UNIX® AND LINUX® SYSTEM ADMINISTRATION HANDBOOK
FIFTH EDITION

Evi Nemeth
Garth Snyder
Trent R. Hein
Ben Whaley
Dan Mackin

with James Garnett, Fabrizio Branca, and Adrian Mouat

▲Addison-Wesley

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  Hot aisles and cold aisles
  Humidity
  Environmental monitoring

Data center reliability tiers

Data center security
  Location
  Perimeter
  Facility access
  Rack access

Tools

Recommended reading

CHAPTER 31: METHODOLOGY, POLICY, AND POLITICS

The grand unified theory: DevOps
  DevOps is CLAMS
  System administration in a DevOps world

Ticketing and task management systems
  Common functions of ticketing systems
  Ticket ownership
  User acceptance of ticketing systems
  Sample ticketing systems
  Ticket dispatching

Local documentation maintenance
  Infrastructure as code
Documentation standards
Environment separation
Disaster management
  Risk assessment
  Recovery planning
  Staffing for a disaster
  Security incidents
IT policies and procedures
  The difference between policies and procedures
  Policy best practices
  Procedures
Service level agreements
  Scope and descriptions of services
  Queue prioritization policies
  Conformance measurements
Compliance: regulations and standards
Legal issues
  Privacy
  Policy enforcement
  Control = liability
  Software licenses
Organizations, conferences, and other resources
Recommended reading

A BRIEF HISTORY OF SYSTEM ADMINISTRATION
COLOPHON
ABOUT THE CONTRIBUTORS
ABOUT THE AUTHORS
INDEX
Every field has an avatar who defines and embodies that space. For system administration, that person is Evi Nemeth.

This is the 5th edition of a book that Evi led as an author for almost three decades. Although Evi wasn’t able to physically join us in writing this edition, she’s with us in spirit and, in some cases, in the form of text and examples that have endured. We’ve gone to great efforts to maintain Evi’s extraordinary style, candor, technical depth, and attention to detail.

An accomplished mathematician and cryptographer, Evi’s professional days were spent (most recently) as a computer science professor at the University of Colorado at Boulder. How system administration came into being, and Evi’s involvement in it, is detailed in the last chapter of this book, *A Brief History of System Administration*.

Throughout her career, Evi looked forward to retiring and sailing the world. In 2001, she did exactly that: she bought a sailboat *(Wonderland)* and set off on an adventure. Across the years, Evi kept us entertained with stories of amazing islands, cool new people, and other sailing escapades. We produced two editions of this book with Evi anchoring as close as possible to shoreline establishments so that she could camp on their Wi-Fi networks and upload chapter drafts.

Never one to decline an intriguing venture, Evi signed on in June 2013 as crew for the historic schooner *Nina* for a sail across the Tasman Sea. The *Nina* disappeared shortly thereafter in a bad storm, and we haven’t heard from Evi since. She was living her dream.

Evi taught us much more than system administration. Even in her 70s, she ran circles around all of us. She was always the best at building a network, configuring a server, debugging a kernel, splitting wood, frying chicken, baking a quiche, or quaffing an occasional glass of wine. With Evi by your side, anything was achievable.

It’s impossible to encapsulate all of Evi’s wisdom here, but these tenets have stuck with us:
• Be conservative in what you send and liberal in what you receive. (This tenet is also known as Postel’s Law, named in honor of Jon Postel, who served as Editor of the RFC series from 1969 until his death in 1998.)
• Be liberal in who you hire, but fire early.
• Don’t use weasel words.
• Undergraduates are the secret superpower.
• You can never use too much red ink.
• You don’t really understand something until you’ve implemented it.
• It’s always time for sushi.
• Be willing to try something twice.
• Always use **sudo**.

We’re sure some readers will write in to ask what, exactly, some of the guidance above really means. We’ve left that as an exercise for the reader, as Evi would have. You can hear her behind you now, saying “Try it yourself. See how it works.”

Smooth sailing, Evi. We miss you.
Preface

Modern technologists are masters at the art of searching Google for answers. If another system administrator has already encountered (and possibly solved) a problem, chances are you can find their write-up on the Internet. We applaud and encourage this open sharing of ideas and solutions.

If great information is already available on the Internet, why write another edition of this book? Here’s how this book helps system administrators grow:

- We offer philosophy, guidance, and context for applying technology appropriately. As with the blind men and the elephant, it’s important to understand any given problem space from a variety of angles. Valuable perspectives include background on adjacent disciplines such as security, compliance, DevOps, cloud computing, and software development life cycles.

- We take a hands-on approach. Our purpose is to summarize our collective perspective on system administration and to recommend approaches that stand the test of time. This book contains numerous war stories and a wealth of pragmatic advice.

- This is not a book about how to run UNIX or Linux at home, in your garage, or on your smartphone. Instead, we describe the management of production environments such as businesses, government offices, and universities. These environments have requirements that are different from (and far outstrip) those of a typical hobbyist.

- We teach you how to be a professional. Effective system administration requires both technical and “soft” skills. It also requires a sense of humor.

The organization of this book

This book is divided into four large chunks: Basic Administration, Networking, Storage, and Operations.
Basic Administration presents a broad overview of UNIX and Linux from a system administrator’s perspective. The chapters in this section cover most of the facts and techniques needed to run a stand-alone system.

The Networking section describes the protocols used on UNIX systems and the techniques used to set up, extend, and maintain networks and Internet-facing servers. High-level network software is also covered here. Among the featured topics are the Domain Name System, electronic mail, single sign-on, and web hosting.

The Storage section tackles the challenges of storing and managing data. This section also covers subsystems that allow file sharing on a network, such as the Network File System and the Windows-friendly SMB protocol.

The Operations section addresses the key topics that a system administrator faces on a daily basis when managing production environments. These topics include monitoring, security, performance, interactions with developers, and the politics of running a system administration group.

Our Contributors

We’re delighted to welcome James Garnett, Fabrizio Branca, and Adrian Mouat as contributing authors for this edition. These contributors’ deep knowledge of a variety of areas has greatly enriched the content of this book.

Contact Information

Please send suggestions, comments, and bug reports to ulsah@book.admin.com. We do answer mail, but please be patient; it is sometimes a few days before one of us is able to respond. Because of the volume of email that this alias receives, we regret that we are unable to answer technical questions.

To view a copy of our current bug list and other late-breaking information, visit our web site, admin.com.

We hope you enjoy this book, and we wish you the best of luck with your adventures in system administration!

Garth Snyder
Trent R. Hein
Ben Whaley
Dan Mackin
July 2017
In 1942, Winston Churchill described an early battle of WWII: “this is not the end—it is not even the beginning of the end—but it is, perhaps, the end of the beginning.” I was reminded of these words when I was approached to write this Foreword for the fifth edition of *UNIX and Linux System Administration Handbook*. The loss at sea of Evi Nemeth has been a great sadness for the UNIX community, but I’m pleased to see her legacy endure in the form of this book and in her many contributions to the field of system administration.

The way the world got its Internet was, originally, through UNIX. A remarkable departure from the complex and proprietary operating systems of its day, UNIX was minimalistic, tools-driven, portable, and widely used by people who wanted to share their work with others. What we today call open source software was already pervasive—but nameless—in the early days of UNIX and the Internet. Open source was just how the technical and academic communities did things, because the benefits so obviously outweighed the costs.

Detailed histories of UNIX, Linux, and the Internet have been lovingly presented elsewhere. I bring up these high-level touchpoints only to remind us all that the modern world owes much to open source software and to the Internet, and that the original foundation for this bounty was UNIX.

As early UNIX and Internet companies fought to hire the most brilliant people and to deliver the most innovative features, software portability was often sacrificed. Eventually, system administrators had to know a little bit about a lot of things because no two UNIX-style operating systems (then, or now) were entirely alike. As a working UNIX system administrator in the mid-1980s and later, I had to know not just shell scripting and Sendmail configuration but also kernel device drivers. It was also important to know how to fix a filesystem with an octal debugger. Fun times!

Out of that era came the first edition of this book and all the editions that followed it. In the parlance of the times, we called the authors “Evi and crew” or perhaps “Evi and her kids.” Because of my work on Cron and BIND, Evi spent a week or two with me (and my family, and my workplace) every time an edition of this book was in progress to make sure she was saying
enough, saying nothing wrong, and hopefully, saying something unique and useful about each of those programs. Frankly, being around Evi was exhausting, especially when she was curious about something, or on a deadline, or in my case, both. That having been said, I miss Evi terribly and I treasure every memory and every photograph of her.

In the decades of this book’s multiple editions, much has changed. It has been fascinating to watch this book evolve along with UNIX itself. Every new edition omitted some technologies that were no longer interesting or relevant to make room for new topics that were just becoming important to UNIX administrators, or that the authors thought soon would be.

It’s hard to believe that we ever spent dozens of kilowatts of power on truck-sized computers whose capabilities are now dwarfed by an Android smartphone. It’s equally hard to believe that we used to run hundreds or thousands of individual server and desktop computers with now-antiquated technologies like rdist. In those years, various editions of this book helped people like me (and like Evi herself) cope with heterogeneous and sometimes proprietary computers that were each real rather than virtualized, and which each had to be maintained rather than being reinstalled (or in Docker, rebuilt) every time something needed patching or upgrading.

We adapt, or we exit. The “Evi kids” who carry on Evi’s legacy have adapted, and they are back in this fifth edition to tell you what you need to know about how modern UNIX and Linux computers work and how you can make them work the way you want them to. Evi’s loss marks the end of an era, but it’s also sobering to consider how many aspects of system administration have passed into history alongside her. I know dozens of smart and successful technologists who will never dress cables in the back of an equipment rack, hear the tone of a modem, or see an RS-232 cable. This edition is for those whose systems live in the cloud or in virtualized data centers; those whose administrative work largely takes the form of automation and configuration source code; those who collaborate closely with developers, network engineers, compliance officers, and all the other worker bees who inhabit the modern hive.

You hold in your hand the latest, best edition of a book whose birth and evolution have precisely tracked the birth and evolution of the UNIX and Internet community. Evi would be extremely proud of her kids, both because of this book, and because of who they have each turned out to be. I am proud to know them.

Paul Vixie
La Honda, California
June 2017
Acknowledgments

Many people contributed to this project, bestowing everything from technical reviews and constructive suggestions to overall moral support.

The following individuals deserve special thanks for hanging in there with us: Jason Carolan, Randy Else, Steve Gaede, Asif Khan, Sam Leathers, Ned McClain, Beth McElroy, Paul Nelson, Tim O’Reilly, Madhuri Peri, Dave Roth, Peter Sankauskas, Deepak Singh, and Paul Vixie.

Our editor at Pearson, Mark Taub, deserves huge thanks for his wisdom, patient support, and gentle author herding throughout the production of this book. It’s safe to say this edition would not have come to fruition without him.

Mary Lou Nohr has been our relentless behind-the-scenes copy editor for over 20 years. When we started work on this edition, Mary Lou was headed for well-deserved retirement. After a lot of begging and guilt-throwing, she agreed to join us for an encore. (Both Mary Lou Nohr and Evi Nemeth appear on the cover. Can you find them?)

We’ve had a fantastic team of technical reviewers. Three dedicated souls reviewed the entire book: Jonathan Corbet, Pat Parseghian, and Jennine Townsend. We greatly appreciate their tenacity and tactfulness.

This edition’s awesome cartoons and cover were conceived and executed by Lisa Haney. Her portfolio is on-line at lisahaney.com.

Last but not least, special thanks to Laszlo Nemeth for his willingness to support the continuation of this series.
SECTION ONE

BASIC ADMINISTRATION
We’ve designed this book to occupy a specific niche in the vast ecosystem of man pages, blogs, magazines, books, and other reference materials that address the needs of UNIX and Linux system administrators.

First, it’s an orientation guide. It reviews the major administrative systems, identifies the different pieces of each, and explains how they work together. In the many cases where you must choose among various implementations of a concept, we describe the advantages and drawbacks of the most popular options.

Second, it’s a quick-reference handbook that summarizes what you need to know to perform common tasks on a variety of common UNIX and Linux systems. For example, the `ps` command, which shows the status of running processes, supports more than 80 command-line options on Linux systems. But a few combinations of options satisfy the majority of a system administrator’s needs; we summarize them on this page.

Finally, this book focuses on the administration of enterprise servers and networks. That is, serious, professional system administration. It’s easy to set up a single system; harder to keep a distributed, cloud-based platform running smoothly in the face of viral popularity, network partitions, and targeted attacks. We describe techniques and rules of thumb that help you recover systems from adversity, and we help you choose solutions that scale as your empire grows in size, complexity, and heterogeneity.
We don’t claim to do all of this with perfect objectivity, but we think we’ve made our biases fairly clear throughout the text. One of the interesting things about system administration is that reasonable people can have dramatically different notions of what constitutes the most appropriate solution. We offer our subjective opinions to you as raw data. Decide for yourself how much to accept and how much of our comments apply to your environment.
1.1 Essential Duties of a System Administrator

The sections below summarize some of the main tasks that administrators are expected to perform. These duties need not necessarily be carried out by a single person, and at many sites the work is distributed among the members of a team. However, at least one person should understand all the components and ensure that every task is performed correctly.
Controlling access

See Chapters 8, 17, and 23 for information about user account provisioning.

The system administrator creates accounts for new users, removes the accounts of inactive users, and handles all the account-related issues that come up in between (e.g., forgotten passwords and lost key pairs). The process of actually adding and removing accounts is typically automated by a configuration management system or centralized directory service.
Adding hardware

Administrators who work with physical hardware (as opposed to cloud or hosted systems) must install it and configure it to be recognized by the operating system. Hardware support chores might range from the simple task of adding a network interface card to configuring a specialized external storage array.
Automating tasks

See Chapter 7, Scripting and the Shell, for information about scripting and automation.

Using tools to automate repetitive and time-consuming tasks increases your efficiency, reduces the likelihood of errors caused by humans, and improves your ability to respond rapidly to changing requirements. Administrators strive to reduce the amount of manual labor needed to keep systems functioning smoothly. Familiarity with scripting languages and automation tools is a large part of the job.
Overseeing backups

See this page for some tips on performing backups.

Backing up data and restoring it successfully when required are important administrative tasks. Although backups are time consuming and boring, the frequency of real-world disasters is simply too high to allow the job to be disregarded.

Operating systems and some individual software packages provide well-established tools and techniques to facilitate backups. Backups must be executed on a regular schedule and restores must be tested periodically to ensure that they are functioning correctly.
Installing and upgrading software

See Chapter 6 for information about software management.

Software must be selected, installed, and configured, often on a variety of operating systems. As patches and security updates are released, they must be tested, reviewed, and incorporated into the local environment without endangering the stability of production systems.

See Chapter 26 for information about software deployment and continuous delivery.

The term “software delivery” refers to the process of releasing updated versions of software—especially software developed in-house—to downstream users. “Continuous delivery” takes this process to the next level by automatically releasing software to users at a regular cadence as it is developed. Administrators help implement robust delivery processes that meet the requirements of the enterprise.
Monitoring

See Chapter 28 for information about monitoring.

Working around a problem is usually faster than taking the time to document and report it, and users internal to an organization often follow the path of least resistance. External users are more likely to voice their complaints publicly than to open a support inquiry. Administrators can help to prevent both of these outcomes by detecting problems and fixing them before public failures occur.

Some monitoring tasks include ensuring that web services respond quickly and correctly, collecting and analyzing log files, and keeping tabs on the availability of server resources such as disk space. All of these are excellent opportunities for automation, and a slew of open source and commercial monitoring systems can help sysadmins with these tasks.
Troubleshooting

See this page for an introduction to network troubleshooting.

Networked systems fail in unexpected and sometimes spectacular fashion. It’s the administrator’s job to play mechanic by diagnosing problems and calling in subject-matter experts as needed. Finding the source of a problem is often more challenging than resolving it.
Maintaining local documentation

See this page for suggestions regarding documentation.

Administrators choose vendors, write scripts, deploy software, and make many other decisions that may not be immediately obvious or intuitive to others. Thorough and accurate documentation is a blessing for team members who would otherwise need to reverse-engineer a system to resolve problems in the middle of the night. A lovingly crafted network diagram is more useful than many paragraphs of text when describing a design.
Vigilantly monitoring security

See Chapter 27 for more information about security.

Administrators are the first line of defense for protecting network-attached systems. The administrator must implement a security policy and set up procedures to prevent systems from being breached. This responsibility might include only a few basic checks for unauthorized access, or it might involve an elaborate network of traps and auditing programs, depending on the context. System administrators are cautious by nature and are often the primary champions of security across a technical organization.
Tuning performance

See *Chapter 29* for more information about performance.

UNIX and Linux are general purpose operating systems that are well suited to almost any conceivable computing task. Administrators can tailor systems for optimal performance in accord with the needs of users, the available infrastructure, and the services the systems provide. When a server is performing poorly, it is the administrator’s job to investigate its operation and identify areas that need improvement.
Developing site policies

See the sections starting on this page for information about local policy-making.

For legal and compliance reasons, most sites need policies that govern the acceptable use of computer systems, the management and retention of data, the privacy and security of networks and systems, and other areas of regulatory interest. System administrators often help organizations develop sensible policies that meet the letter and intent of the law and yet still promote progress and productivity.
**Working with vendors**

Most sites rely on third parties to provide a variety of ancillary services and products related to their computing infrastructure. These providers might include software developers, cloud infrastructure providers, hosted software-as-a-service (SaaS) shops, help-desk support staff, consultants, contractors, security experts, and platform or infrastructure vendors. Administrators may be tasked with selecting vendors, assisting with contract negotiations, and implementing solutions once the paperwork has been completed.
Fire fighting

Although helping other people with their various problems is rarely included in a system administrator’s job description, these tasks claim a measurable portion of most administrators’ workdays. System administrators are bombarded with problems ranging from “It worked yesterday and now it doesn’t! What did you change?” to “I spilled coffee on my keyboard! Should I pour water on it to wash it out?”

In most cases, your response to these issues affects your perceived value as an administrator far more than does any actual technical skill you might possess. You can either howl at the injustice of it all, or you can delight in the fact that a single well-handled trouble ticket scores more brownie points than five hours of midnight debugging. Your choice!
1.2 SUGGESTED BACKGROUND

We assume in this book that you have a certain amount of Linux or UNIX experience. In particular, you should have a general concept of how the system looks and feels from a user’s perspective since we do not review that material. Several good books can get you up to speed; see Recommended reading.

We love well-designed graphical interfaces. Unfortunately, GUI tools for system administration on UNIX and Linux remain rudimentary in comparison with the richness of the underlying software. In the real world, administrators must be comfortable using the command line.

For text editing, we strongly recommend learning vi (now seen more commonly in its enhanced form, vim), which is standard on all systems. It is simple, powerful, and efficient. Mastering vim is perhaps the single best productivity enhancement available to administrators. Use the vintutor command for an excellent, interactive introduction.

Alternatively, GNU’s nano is a simple and low-impact “starter editor” that has on-screen prompts. Use it discreetly; professional administrators may be visibly distressed if they witness a peer running nano.

See Chapter 7 for an introduction to scripting.

Although administrators are not usually considered software developers, industry trends are blurring the lines between these functions. Capable administrators are usually polyglot programmers who don’t mind picking up a new language when the need arises.

For new scripting projects, we recommend Bash (aka bash, aka sh), Ruby, or Python. Bash is the default command shell on most UNIX and Linux systems. It is primitive as a programming language, but it serves well as the duct tape in an administrative tool box. Python is a clever language with a highly readable syntax, a large developer community, and libraries that facilitate many common tasks. Ruby developers describe the language as “a joy to work with” and “beautiful to behold.” Ruby and Python are similar in many ways, and we’ve found them to be equally functional for administration. The choice between them is mostly a matter of personal preference.

We also suggest that you learn expect, which is not a programming language so much as a front end for driving interactive programs. It’s an efficient glue technology that can replace some complex scripting and is easy to learn.

Chapter 7, Scripting and the Shell, summarizes the most important things to know about scripting for Bash, Python, and Ruby. It also reviews regular expressions (text matching patterns) and some shell idioms that are useful for sysadmins.
1.3 Linux Distributions

A Linux distribution comprises the Linux kernel, which is the core of the operating system, and packages that make up all the commands you can run on the system. All distributions share the same kernel lineage, but the format, type, and number of packages differ quite a bit. Distributions also vary in their focus, support, and popularity. There continue to be hundreds of independent Linux distributions, but our sense is that distributions derived from the Debian and Red Hat lineages will predominate in production environments in the years ahead.

By and large, the differences among Linux distributions are not cosmically significant. In fact, it is something of a mystery why so many different distributions exist, each claiming “easy installation” and “a massive software library” as its distinguishing features. It’s hard to avoid the conclusion that people just like to make new Linux distributions.

Most major distributions include a relatively painless installation procedure, a desktop environment, and some form of package management. You can try them out easily by starting up a cloud instance or a local virtual machine.

See Chapter 25, Containers, for more information about Docker and containers.

Much of the insecurity of general-purpose operating systems derives from their complexity. Virtually all leading distributions are cluttered with scores of unused software packages; security vulnerabilities and administrative anguish often come along for the ride. In response, a relatively new breed of minimalist distributions has been gaining traction. CoreOS is leading the charge against the status quo and prefers to run all software in containers. Alpine Linux is a lightweight distribution that is used as the basis of many public Docker images. Given this reductionist trend, we expect the footprint of Linux to shrink over the coming years.

By adopting a distribution, you are making an investment in a particular vendor’s way of doing things. Instead of looking only at the features of the installed software, it’s wise to consider how your organization and that vendor are going to work with each other. Some important questions to ask are:

- Is this distribution going to be around in five years?
- Is this distribution going to stay on top of the latest security patches?
- Does this distribution have an active community and sufficient documentation?
- If I have problems, will the vendor talk to me, and how much will that cost?

Table 1.1 lists some of the most popular mainstream distributions.
**Table 1.1: Most popular general-purpose Linux distributions**

<table>
<thead>
<tr>
<th>Distribution</th>
<th>Web site</th>
<th>Comments</th>
</tr>
</thead>
<tbody>
<tr>
<td>Arch</td>
<td>archlinux.org</td>
<td>For those who fear not the command line</td>
</tr>
<tr>
<td>CentOS</td>
<td>centos.org</td>
<td>Free analog of Red Hat Enterprise</td>
</tr>
<tr>
<td>CoreOS</td>
<td>coreos.com</td>
<td>Containers, containers everywhere</td>
</tr>
<tr>
<td>Debian</td>
<td>debian.org</td>
<td>Free as in freedom, most GNUish distro</td>
</tr>
<tr>
<td>Fedora</td>
<td>fedoraproject.org</td>
<td>Test bed for Red Hat Linux</td>
</tr>
<tr>
<td>Kali</td>
<td>kali.org</td>
<td>For penetration testers</td>
</tr>
<tr>
<td>Linux Mint</td>
<td>linuxmint.com</td>
<td>Ubuntu-based, desktop-friendly</td>
</tr>
<tr>
<td>openSUSE</td>
<td>opensuse.org</td>
<td>Free analog of SUSE Linux Enterprise</td>
</tr>
<tr>
<td>openWRT</td>
<td>openwrt.org</td>
<td>Linux for routers and embedded devices</td>
</tr>
<tr>
<td>Oracle Linux</td>
<td>oracle.com</td>
<td>Oracle-supported version of RHEL</td>
</tr>
<tr>
<td>RancherOS</td>
<td>rancher.com</td>
<td>20MiB, everything in containers</td>
</tr>
<tr>
<td>Red Hat Enterprise</td>
<td>redhat.com</td>
<td>Reliable, slow-changing, commercial</td>
</tr>
<tr>
<td>Slackware</td>
<td>slackware.com</td>
<td>Grizzled, long-surviving distro</td>
</tr>
<tr>
<td>SUSE Linux Enterprise</td>
<td>suse.com</td>
<td>Strong in Europe, multilingual</td>
</tr>
<tr>
<td>Ubuntu</td>
<td>ubuntu.com</td>
<td>Cleaned-up version of Debian</td>
</tr>
</tbody>
</table>

The most viable distributions are not necessarily the most corporate. For example, we expect Debian Linux (OK, OK, Debian GNU/Linux!) to remain viable for a long time despite the fact that Debian is not a company, doesn’t sell anything, and offers no enterprise-level support. Debian benefits from a committed group of contributors and from the enormous popularity of the Ubuntu distribution, which is based on it.

A comprehensive list of distributions, including many non-English distributions, can be found at [lwn.net/Distributions](http://lwn.net/Distributions) or distrowatch.com.
1.4 Example Systems Used in This Book

We have chosen three popular Linux distributions and one UNIX variant as our primary examples for this book: Debian GNU/Linux, Ubuntu Linux, Red Hat Enterprise Linux (and its dopplegänger CentOS), and FreeBSD. These systems are representative of the overall marketplace and account collectively for a substantial portion of installations in use at large sites today.

Information in this book generally applies to all of our example systems unless a specific attribution is given. Details particular to one system are marked with a logo:

- Debian GNU/Linux 9.0 “Stretch”
- Ubuntu® 17.04 “Zesty Zapus”
- Red Hat® Enterprise Linux® 7.1 and CentOS® 7.1
- FreeBSD® 11.0

Most of these marks belong to the vendors that release the corresponding software and are used with the kind permission of their respective owners. However, the vendors have not reviewed or endorsed the contents of this book.

We repeatedly attempted and failed to obtain permission from Red Hat to use their famous red fedora logo, so you’re stuck with yet another technical acronym. At least this one is in the margins.

The paragraphs below provide a bit more detail about each of the example systems.
Example Linux distributions

Information that’s specific to Linux but not to any particular distribution is marked with the Tux penguin logo shown at left.

Debian (pronounced deb-ian, named after the late founder Ian Murdock and his wife Debra), is one of the oldest and most well-regarded distributions. It is a noncommercial project with more than a thousand contributors worldwide. Debian maintains an ideological commitment to community development and open access, so there’s never any question about which parts of the distribution are free or redistributable.

Debian defines three releases that are maintained simultaneously: stable, targeting production servers; unstable, with current packages that may have bugs and security vulnerabilities; and testing, which is somewhere in between.

Ubuntu is based on Debian and maintains Debian’s commitment to free and open source software. The business behind Ubuntu is Canonical Ltd., founded by entrepreneur Mark Shuttleworth.

Canonical offers a variety of editions of Ubuntu targeting the cloud, the desktop, and bare metal. There are even releases intended for phones and tablets. Ubuntu version numbers derive from the year and month of release, so version 16.10 is from October, 2016. Each release also has an alliterative code name such as Vivid Vervet or Wily Werewolf.

Two versions of Ubuntu are released annually: one in April and one in October. The April releases in even-numbered years are long-term support (LTS) editions that promise five years of maintenance updates. These are the releases recommended for production use.

RHEL

Red Hat has been a dominant force in the Linux world for more than two decades, and its distributions are widely used in North America and beyond. By the numbers, Red Hat, Inc., is the most successful open source software company in the world.

Red Hat Enterprise Linux, often shortened to RHEL, targets production environments at large enterprises that require support and consulting services to keep their systems running smoothly. Somewhat paradoxically, RHEL is open source but requires a license. If you’re not willing to pay for the license, you’re not going to be running Red Hat.

Red Hat also sponsors Fedora, a community-based distribution that serves as an incubator for bleeding-edge software not considered stable enough for RHEL. Fedora is used as the initial test bed for software and configurations that later find their way to RHEL.

CentOS is virtually identical to Red Hat Enterprise Linux, but free of charge. The CentOS Project (centos.org) is owned by Red Hat and employs its lead developers. However, they operate separately from the Red Hat Enterprise Linux team. The CentOS distribution lacks Red Hat’s branding and a few proprietary tools, but is in other respects equivalent.
CentOS is an excellent choice for sites that want to deploy a production-oriented distribution without paying tithes to Red Hat. A hybrid approach is also feasible: front-line servers can run Red Hat Enterprise Linux and avail themselves of Red Hat’s excellent support, even as nonproduction systems run CentOS. This arrangement covers the important bases in terms of risk and support while also minimizing cost and administrative complexity.

CentOS aspires to full binary and bug-for-bug compatibility with Red Hat Enterprise Linux. Rather than repeating “Red Hat and CentOS” ad nauseam, we generally mention only one or the other in this book. The text applies equally to Red Hat and CentOS unless we note otherwise.

Other popular distributions are also Red Hat descendants. Oracle sells a rebranded and customized version of CentOS to customers of its enterprise database software. Amazon Linux, available to Amazon Web Services users, was initially derived from CentOS and still shares many of its conventions.

Most administrators will encounter a Red Hat-like system at some point in their careers, and familiarity with its nuances is helpful even if it isn’t the system of choice at your site.
Example UNIX distribution

The popularity of UNIX has been waning for some time, and most of the stalwart UNIX distributions (e.g., Solaris, HP-UX, and AIX) are no longer in common use. The open source descendants of BSD are exceptions to this trend and continue to enjoy a cult following, particularly among operating system experts, free software evangelists, and security-minded administrators. In other words, some of the world’s foremost operating system authorities rely on the various BSD distributions. Apple’s macOS has a BSD heritage.

FreeBSD, first released in late 1993, is the most widely used of the BSD derivatives. It commands a 70% market share among BSD variants according to some usage statistics. Users include major Internet companies such as WhatsApp, Google, and Netflix.

Unlike Linux, FreeBSD is a complete operating system, not just a kernel. Both the kernel and userland software are licensed under the permissive BSD License, a fact that encourages development by and additions from the business community.
1.5 **Notation and Typographical Conventions**

In this book, filenames, commands, and literal arguments to commands are shown in boldface. Placeholders (e.g., command arguments that should not be taken literally) are in italics. For example, in the command

\texttt{cp file directory}

you’re supposed to replace \textit{file} and \textit{directory} with the names of an actual file and an actual directory.

Excerpts from configuration files and terminal sessions are shown in a code font. Sometimes, we annotate sessions with the \texttt{bash} comment character \# and italic text. For example:

\begin{verbatim}
$ grep Bob /pub/phonelist  # Look up Bob's phone number
Bob Knowles 555-2834
Bob Smith 555-2311
\end{verbatim}

We use $ to denote the shell prompt for a normal, unprivileged user, and # for the root user. When a command is specific to a distribution or family of distributions, we prefix the prompt with the distribution name. For example:

\begin{verbatim}
$ sudo su - root  # Become root
# passwd         # Change root's password
debian# dpkg -l  # List installed packages on Debian and Ubuntu
\end{verbatim}

This convention is aligned with the one used by standard UNIX and Linux shells.

Outside of these specific cases, we have tried to keep special fonts and formatting conventions to a minimum as long as we could do so without compromising intelligibility. For example, we often talk about entities such as the daemon group with no special formatting at all.

We use the same conventions as the manual pages for command syntax:

- Anything between square brackets ("[" and "]") is optional.
- Anything followed by an ellipsis ("…") can be repeated.
- Curly braces ("{" and "}") mean that you should select one of the items separated by vertical bars ("|").

For example, the specification

\texttt{bork \ [-x] \ [ \ on | off \] \ filename \ ...}

would match any of the following commands:
We use shell-style globbing characters for pattern matching:

- A star (*) matches zero or more characters.
- A question mark (?) matches one character.
- A tilde or “twiddle” (~) means the home directory of the current user.
- ~user means the home directory of user.

For example, we might refer to the startup script directories /etc/rc0.d, /etc/rc1.d, and so on with the shorthand pattern /etc/rc*.d.

Text within quotation marks often has a precise technical meaning. In these cases, we ignore the normal rules of U.S. English and put punctuation outside the quotes so that there can be no confusion about what’s included and what’s not.
1.6 Units

Metric prefixes such as kilo-, mega-, and giga- are defined as powers of 10; one megabuck is $1,000,000. However, computer types have long poached these prefixes and used them to refer to powers of 2. For example, one “megabyte” of memory is really $2^{20}$ or 1,048,576 bytes. The stolen units have even made their way into formal standards such as the JEDEC Solid State Technology Association’s Standard 100B.01, which recognizes the prefixes as denoting powers of 2 (albeit with some misgivings).

In an attempt to restore clarity, the International Electrotechnical Commission has defined a set of numeric prefixes (kibi-, mebi-, gibi-, and so on, abbreviated Ki, Mi, and Gi) based explicitly on powers of 2. Those units are always unambiguous, but they are just starting to be widely used. The original kilo-series prefixes are still used in both senses.

Context helps with decoding. RAM is always denominated in powers of 2, but network bandwidth is always a power of 10. Storage space is usually quoted in power-of-10 units, but block and page sizes are in fact powers of 2.

In this book, we use IEC units for powers of 2, metric units for powers of 10, and metric units for rough values and cases in which the exact basis is unclear, undocumented, or impossible to determine. In command output and in excerpts from configuration files, or where the delineation is not important, we leave the original values and unit designators. We abbreviate bit as b and byte as B. Table 1.2 shows some examples.

<table>
<thead>
<tr>
<th>Example</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>1kB file</td>
<td>A file that contains 1,000 bytes</td>
</tr>
<tr>
<td>4KiB SSD pages</td>
<td>SSD pages that contain 4,096 bytes</td>
</tr>
<tr>
<td>8KB of memory</td>
<td>Not used in this book; see note below</td>
</tr>
<tr>
<td>100MB file size limit</td>
<td>Nominally $10^8$ bytes; in context, ambiguous</td>
</tr>
<tr>
<td>100MB disk partition</td>
<td>Nominally $10^8$ bytes; in context, probably 99,999,744 bytes*</td>
</tr>
<tr>
<td>1GiB of RAM</td>
<td>1,073,741,824 bytes of memory</td>
</tr>
<tr>
<td>1 Gb/s Ethernet</td>
<td>A network that transmits 1,000,000,000 bits per second</td>
</tr>
<tr>
<td>6TB hard disk</td>
<td>A hard disk that stores about 6,000,000,000,000 bytes</td>
</tr>
</tbody>
</table>

* That is, $10^8$ rounded down to the nearest whole multiple of the disk’s 512-byte block size

The abbreviation K, as in “8KB of RAM!”’, is not part of any standard. It’s a computerese adaptation of the metric abbreviation k, for kilo-, and originally meant 1,024 as opposed to 1,000. But since the abbreviations for the larger metric prefixes are already upper case, the analogy doesn’t scale. Later, people became confused about the distinction and started using K for factors of 1,000, too.
Most of the world doesn’t consider this to be an important matter and, like the use of imperial units in the United States, metric prefixes are likely to be misused for the foreseeable future. Ubuntu maintains a helpful units policy, though we suspect it has not been widely adopted even at Canonical; see wiki.ubuntu.com/UnitsPolicy for some additional details.
1.7 Man pages and other on-line documentation

The manual pages, usually called “man pages” because they are read with the `man` command, constitute the traditional “on-line” documentation. (Of course, these days all documentation is on-line in some form or another.) Program-specific man pages come along for the ride when you install new software packages. Even in the age of Google, we continue to consult man pages as an authoritative resource because they are accessible from the command line, typically include complete details on a program’s options, and show helpful examples and related commands.

Man pages are concise descriptions of individual commands, drivers, file formats, or library routines. They do not address more general topics such as “How do I install a new device?” or “Why is this system so damn slow?”
Organization of the man pages

FreeBSD and Linux divide the man pages into sections. Table 1.3 shows the basic schema. Other UNIX variants sometimes define the sections slightly differently.

Table 1.3: Sections of the man pages

<table>
<thead>
<tr>
<th>Section</th>
<th>Contents</th>
</tr>
</thead>
<tbody>
<tr>
<td>1</td>
<td>User-level commands and applications</td>
</tr>
<tr>
<td>2</td>
<td>System calls and kernel error codes</td>
</tr>
<tr>
<td>3</td>
<td>Library calls</td>
</tr>
<tr>
<td>4</td>
<td>Device drivers and network protocols</td>
</tr>
<tr>
<td>5</td>
<td>Standard file formats</td>
</tr>
<tr>
<td>6</td>
<td>Games and demonstrations</td>
</tr>
<tr>
<td>7</td>
<td>Miscellaneous files and documents</td>
</tr>
<tr>
<td>8</td>
<td>System administration commands</td>
</tr>
<tr>
<td>9</td>
<td>Obscure kernel specs and interfaces</td>
</tr>
</tbody>
</table>

The exact structure of the sections isn’t important for most topics because man finds the appropriate page wherever it is stored. Just be aware of the section definitions when a topic with the same name appears in multiple sections. For example, passwd is both a command and a configuration file, so it has entries in both section 1 and section 5.
**man: read man pages**

*man title* formats a specific manual page and sends it to your terminal through *more*, *less*, or whatever program is specified in your PAGER environment variable. *title* is usually a command, device, filename, or name of a library routine. The sections of the manual are searched in roughly numeric order, although sections that describe commands (sections 1 and 8) are usually searched first.

---

See [this page](#) to learn about environment variables.

The form *man section title* gets you a man page from a particular section. Thus, on most systems, *man sync* gets you the man page for the *sync* command, and *man 2 sync* gets you the man page for the *sync* system call.

*man -k keyword* or *apropos keyword* prints a list of man pages that have *keyword* in their one-line synopses. For example:

```
$ man -k translate
objcopy (1)   - copy and translate object files
dgettext (3)  - translate message
tr (1)        - translate or delete characters
snmptranslate (1) - translate SNMP OID values into useful information
tr (1p)       - translate characters
...
```

The keywords database can become outdated. If you add additional man pages to your system, you may need to rebuild this file with *makewhatis* (Red Hat and FreeBSD) or *mandb* (Ubuntu).
Storage of man pages

nroff input for man pages (i.e., the man page source code) is stored in directories under /usr/share/man and compressed with gzip to save space. The man command knows how to decompress them on the fly.

man maintains a cache of formatted pages in /var/cache/man or /usr/share/man if the appropriate directories are writable; however, this is a security risk. Most systems preformat the man pages once at installation time (see catman) or not at all.

The man command can search several man page repositories to find the manual pages you request. On Linux systems, you can find out the current default search path with the manpath command. This path (from Ubuntu) is typical:

```
ubuntu$ manpath
/usr/local/man:/usr/local/share/man:/usr/share/man
```

If necessary, you can set your MANPATH environment variable to override the default path:

```
$ export MANPATH=/home/share/localman:/usr/share/man
```

Some systems let you set a custom system-wide default search path for man pages, which can be useful if you need to maintain a parallel tree of man pages such as those generated by OpenPKG. To distribute local documentation in the form of man pages, however, it is simpler to use your system’s standard packaging mechanism and to put man pages in the standard man directories. See Chapter 6, Software Installation and Management, for more details.
1.8 **Other authoritative documentation**

Man pages are just a small part of the official documentation. Most of the rest, unfortunately, is scattered about on the web.
System-specific guides

Major vendors have their own dedicated documentation projects. Many continue to produce useful book-length manuals, including administration and installation guides. These are generally available on-line and as downloadable PDF files. Table 1.4 shows where to look.

Table 1.4: Where to find OS vendors’ proprietary documentation

<table>
<thead>
<tr>
<th>OS</th>
<th>URL</th>
<th>Comments</th>
</tr>
</thead>
<tbody>
<tr>
<td>Debian</td>
<td>debian.org/doc</td>
<td>Admin handbook lags behind the current version</td>
</tr>
<tr>
<td>Ubuntu</td>
<td>help.ubuntu.com</td>
<td>User oriented, see “server guide” for LTS releases</td>
</tr>
<tr>
<td>RHEL</td>
<td>redhat.com/docs</td>
<td>Comprehensive docs for administrators</td>
</tr>
<tr>
<td>CentOS</td>
<td>wiki.centos.org</td>
<td>Includes tips, HowTos, and FAQs</td>
</tr>
<tr>
<td>FreeBSD</td>
<td>freebsd.org/docs.html</td>
<td>See the FreeBSD Handbook for sysadmin info</td>
</tr>
</tbody>
</table>

Although this documentation is helpful, it’s not the sort of thing you keep next to your bed for light evening reading (though some vendors’ versions would make useful sleep aids). We generally Google for answers before turning to vendor docs.
Package-specific documentation

Most of the important software packages in the UNIX and Linux world are maintained by individuals or by third parties such as the Internet Systems Consortium and the Apache Software Foundation. These groups write their own documentation. The quality runs the gamut from embarrassing to spectacular, but jewels such as *Pro Git* from [git-scm.com/book](http://git-scm.com/book) make the hunt worthwhile.

Supplemental documents include white papers (technical reports), design rationales, and book- or pamphlet-length treatments of particular topics. These supplemental materials are not limited to describing just one command, so they can adopt a tutorial or procedural approach. Many pieces of software have both a man page and a long-form article. For example, the man page for **vim** tells you about the command-line arguments that **vim** understands, but you have to turn to an in-depth treatment to learn how to actually edit a file.

Most software projects have user and developer mailing lists and IRC channels. This is the first place to visit if you have questions about a specific configuration issue or if you encounter a bug.
Books

The O’Reilly books are favorites in the technology industry. The business began with *UNIX in a Nutshell* and now includes a separate volume on just about every important UNIX and Linux subsystem and command. O’Reilly also publishes books on network protocols, programming languages, Microsoft Windows, and other non-UNIX tech topics. All the books are reasonably priced, timely, and focused.

Many readers turn to O’Reilly’s Safari Books Online, a subscription service that offers unlimited electronic access to books, videos, and other learning resources. Content from many publishers is included—not just O’Reilly—and you can choose from an immense library of material.
RFC publications

Request for Comments documents describe the protocols and procedures used on the Internet. Most of these are relatively detailed and technical, but some are written as overviews. The phrase “reference implementation” applied to software usually translates to “implemented by a trusted source according to the RFC specification.”

RFCs are absolutely authoritative, and many are quite useful for system administrators. See this page for a more complete description of these documents. We refer to various RFCs throughout this book.
1.9 Other sources of information

The sources discussed in the previous section are peer reviewed and written by authoritative sources, but they’re hardly the last word in UNIX and Linux administration. Countless blogs, discussion forums, and news feeds are available on the Internet.

It should go without saying, but Google is a system administrator’s best friend. Unless you’re looking up the details of a specific command or file format, Google or an equivalent search engine should be the first resource you consult for any sysadmin question. Make it a habit; if nothing else, you’ll avoid the delay and humiliation of having your questions in an on-line forum answered with a link to Google (or worse yet, a link to Google through lmgtfy.com). When stuck, search the web.
Keeping current

Operating systems and the tools and techniques that support them change rapidly. Read the sites in Table 1.5 with your morning coffee to keep abreast of industry trends.

Table 1.5: Resources for keeping up to date

<table>
<thead>
<tr>
<th>Web site</th>
<th>Description</th>
</tr>
</thead>
<tbody>
<tr>
<td>darkreading.com</td>
<td>Security news, trends, and discussion</td>
</tr>
<tr>
<td>devopsreactions.tumblr.com</td>
<td>Sysadmin humor in animated GIF form</td>
</tr>
<tr>
<td>linux.com</td>
<td>A Linux Foundation site; forum, good for new users</td>
</tr>
<tr>
<td>linuxfoundation.org</td>
<td>Nonprofit fostering OSS, employer of Linus Torvalds</td>
</tr>
<tr>
<td>lwn.net</td>
<td>High-quality, timely articles on Linux and OSS</td>
</tr>
<tr>
<td>lxer.com</td>
<td>Linux news aggregator</td>
</tr>
<tr>
<td>securityfocus.com</td>
<td>Vulnerability reports and security-related mailing lists</td>
</tr>
<tr>
<td>@SwiftOnSecurity</td>
<td>Infosec opinion from Taylor Swift (parody account)</td>
</tr>
<tr>
<td>@nixcraft</td>
<td>Tweets about UNIX and Linux administration</td>
</tr>
<tr>
<td>everythingsysadmin.com</td>
<td>Blog of Thomas Limoncelli, respected sysadmin</td>
</tr>
<tr>
<td>sysadvent.blogspot.com</td>
<td>Advent for sysadmins with articles each December</td>
</tr>
<tr>
<td>oreilly.com/topics</td>
<td>Learning resources from O'Reilly on many topics</td>
</tr>
<tr>
<td>schneier.com</td>
<td>Blog of Bruce Schneier, privacy and security expert</td>
</tr>
</tbody>
</table>

a. See also Tom’s collection of April Fools’ Day RFCs at rfc-humor.com

Social media are also useful. Twitter and reddit in particular have strong, engaged communities with a lot to offer, though the signal-to-noise ratio can sometimes be quite bad. On reddit, join the sysadmin, linux, linuxadmin, and netsec subreddits.
HowTos and reference sites

The sites listed in Table 1.6 contain guides, tutorials, and articles about how to accomplish specific tasks on UNIX and Linux.

Table 1.6: Task-specific forums and reference sites

<table>
<thead>
<tr>
<th>Web site</th>
<th>Description</th>
</tr>
</thead>
<tbody>
<tr>
<td>wiki.archlinux.org</td>
<td>Articles and guides for Arch Linux; many are more general</td>
</tr>
<tr>
<td>askubuntu.com</td>
<td>Q&amp;A for Ubuntu users and developers</td>
</tr>
<tr>
<td>digitalocean.com</td>
<td>Tutorials on many OSS, development, and sysadmin topics&lt;sup&gt;a&lt;/sup&gt;</td>
</tr>
<tr>
<td>kernel.org</td>
<td>Official Linux kernel site</td>
</tr>
<tr>
<td>serverfault.com</td>
<td>Collaboratively edited database of sysadmin questions&lt;sup&gt;b&lt;/sup&gt;</td>
</tr>
<tr>
<td>serversforhackers.com</td>
<td>High-quality videos, forums, and articles on administration</td>
</tr>
</tbody>
</table>

<sup>a</sup> See digitalocean.com/community/tutorials
<sup>b</sup> Also see the sister site stackoverflow.com, which is dedicated to programming but useful for sysadmins

Stack Overflow and Server Fault, both listed in Table 1.6 (and both members of the Stack Exchange group of sites), warrant a closer look. If you’re having a problem, chances are that somebody else has already seen it and asked for help on one of these sites. The reputation-based Q&A format used by the Stack Exchange sites has proved well suited to the kinds of problems that sysadmins and programmers encounter. It’s worth creating an account and joining this large community.
Conferences

Industry conferences are a great way to network with other professionals, keep tabs on technology trends, take training classes, gain certifications, and learn about the latest services and products. The number of conferences pertinent to administration has exploded in recent years. Table 1.7 highlights some of the most prominent ones.

Table 1.7: Conferences relevant to system administrators

<table>
<thead>
<tr>
<th>Conference</th>
<th>Location</th>
<th>When</th>
<th>Description</th>
</tr>
</thead>
<tbody>
<tr>
<td>LISA</td>
<td>Varies</td>
<td>Q4</td>
<td>Large Installation System Administration</td>
</tr>
<tr>
<td>Monitorama</td>
<td>Portland</td>
<td>June</td>
<td>Monitoring tools and techniques</td>
</tr>
<tr>
<td>OSCON</td>
<td>Varies (US/EU)</td>
<td>Q2 or Q3</td>
<td>Long-running O’Reilly OSS conference</td>
</tr>
<tr>
<td>SCALE</td>
<td>Pasadena</td>
<td>Jan</td>
<td>Southern California Linux Expo</td>
</tr>
<tr>
<td>DefCon</td>
<td>Las Vegas</td>
<td>July</td>
<td>Oldest and largest hacker convention</td>
</tr>
<tr>
<td>Velocity</td>
<td>Global</td>
<td>Varies</td>
<td>O’Reilly conference on web operations</td>
</tr>
<tr>
<td>BSDCan</td>
<td>Ottawa</td>
<td>May/June</td>
<td>Everything BSD from novices to gurus</td>
</tr>
<tr>
<td>re:Invent</td>
<td>Las Vegas</td>
<td>Q4</td>
<td>AWS cloud computing conference</td>
</tr>
<tr>
<td>VMWorld</td>
<td>Varies (US/EU)</td>
<td>Q3 or Q4</td>
<td>Virtualization and cloud computing</td>
</tr>
<tr>
<td>LinuxCon</td>
<td>Global</td>
<td>Varies</td>
<td>The future of Linux</td>
</tr>
<tr>
<td>RSA</td>
<td>San Francisco</td>
<td>Q1 or Q2</td>
<td>Enterprise cryptography and infosec</td>
</tr>
<tr>
<td>DevOpsDays</td>
<td>Global</td>
<td>Varies</td>
<td>A range of topics on bridging the gap between development and ops teams</td>
</tr>
<tr>
<td>QCon</td>
<td>Global</td>
<td>Varies</td>
<td>A conference for software developers</td>
</tr>
</tbody>
</table>

Meetups (meetup.com) are another way to network and engage with like-minded people. Most urban areas in the United States and around the world have a Linux user group or DevOps meetup that sponsors speakers, discussions, and hack days.
1.10 Ways to Find and Install Software

Chapter 6, Software Installation and Management, addresses software provisioning in detail. But for the impatient, here’s a quick primer on how to find out what’s installed on your system and how to obtain and install new software.

Modern operating systems divide their contents into packages that can be installed independently of one another. The default installation includes a range of starter packages that you can expand and contract according to your needs. When adding software, don your security hat and remember that additional software creates additional attack surface. Only install what’s necessary.

Add-on software is often provided in the form of precompiled packages as well, although the degree to which this is a mainstream approach varies widely among systems. Most software is developed by independent groups that release the software in the form of source code. Package repositories then pick up the source code, compile it appropriately for the conventions in use on the systems they serve, and package the resulting binaries. It’s usually easier to install a system-specific binary package than to fetch and compile the original source code. However, packagers are sometimes a release or two behind the current version.

The fact that two systems use the same package format doesn’t necessarily mean that packages for the two systems are interchangeable. Red Hat and SUSE both use RPM, for example, but their filesystem layouts are somewhat different. It’s best to use packages designed for your particular system if they are available.

Our example systems provide excellent package management systems that include tools for accessing and searching hosted software repositories. Distributors aggressively maintain these repositories on behalf of the community, to facilitate patching and software updates. Life is good.

When the packaged format is insufficient, administrators must install software the old-fashioned way: by downloading a tar archive of the source code and manually configuring, compiling, and installing it. Depending on the software and the operating system, this process can range from trivial to nightmarish.

In this book, we generally assume that optional software is already installed rather than torturing you with boilerplate instructions for installing every package. If there’s a potential for confusion, we sometimes mention the exact names of the packages needed to complete a particular project. For the most part, however, we don’t repeat installation instructions since they tend to be similar from one package to the next.
Determining if software is already installed

For a variety of reasons, it can be a bit tricky to determine which package contains the software you actually need. Rather than starting at the package level, it’s easier to use the shell’s `which` command to find out if a relevant binary is already in your search path. For example, the following command reveals that the GNU C compiler has already been installed on this machine:

```
ubuntu$ which gcc
/usr/bin/gcc
```

If `which` can’t find the command you’re looking for, try `whereis`; it searches a broader range of system directories and is independent of your shell’s search path.

Another alternative is the incredibly useful `locate` command, which consults a precompiled index of the filesystem to locate filenames that match a particular pattern.

FreeBSD includes `locate` as part of the base system. In Linux, the current implementation of `locate` is in the `mlocate` package. On Red Hat and CentOS, install the `mlocate` package with `yum`; see this page.

`locate` can find any type of file; it is not specific to commands or packages. For example, if you weren’t sure where to find the `signal.h` include file, you could try

```
freebsd$ locate signal.h
/usr/include/machine/signal.h
/usr/include/signal.h
/usr/include/sys/signal.h
...
```

`locate`’s database is updated periodically by the `updatedb` command (in FreeBSD, `locate.updatedb`), which runs periodically out of `cron`. Therefore, the results of a `locate` don’t always reflect recent changes to the filesystem.

See Chapter 6 for more information about package management.

If you know the name of the package you’re looking for, you can also use your system’s packaging utilities to check directly for the package’s presence. For example, on a Red Hat system, the following command checks for the presence (and installed version) of the Python interpreter:

```
redhat$ rpm --query python
python-2.7.5-18.el7_1.i.x86_64
```

You can also find out which package a particular file belongs to:
redhat$ rpm -qf /etc/httpd
httpd-2.4.6-31.el7.centos.x86_64

freebsd$ pkg which /usr/local/sbin/httpd
/usr/local/sbin/httpd was installed by package apache24-2.4.12

ubuntu$ dpkg-query -S /etc/apache2
apache2: /etc/apache2
Adding new software

If you do need to install additional software, you first need to determine the canonical name of the relevant software package. For example, you’d need to translate “I want to install locate” to “I need to install the mlocate package,” or translate “I need named” to “I have to install BIND.” A variety of system-specific indexes on the web can help with this, but Google is usually just as effective. For example, a search for “locate command” takes you directly to several relevant discussions.

The following examples show the installation of the tcpdump command on each of our example systems. tcpdump is a packet capture tool that lets you view the raw packets being sent to and from the system on the network.

Debian and Ubuntu use APT, the Debian Advanced Package Tool:

```bash
ubuntu# sudo apt-get install tcpdump
Reading package lists... Done
Building dependency tree
Reading state information... Done
The following NEW packages will be installed:
  tcpdump
0 upgraded, 1 newly installed, 0 to remove and 81 not upgraded.
Need to get 0 B/360 kB of archives.
After this operation, 1,179 kB of additional disk space will be used.
Selecting previously unselected package tcpdump.
(Reading database ... 63846 files and directories currently installed.)
Preparing to unpack .../tcpdump_4.6.2-4ubuntu1_amd64.deb ...
Unpacking tcpdump (4.6.2-4ubuntu1) ...
Processing triggers for man-db (2.7.0.2-5) ...
Setting up tcpdump (4.6.2-4ubuntu1) ...
```

The Red Hat and CentOS version is
redhat# sudo yum install tcpdump
Loaded plugins: fastestmirror
Determining fastest mirrors
  * base: mirrors.xmission.com
  * epel: linux.mirrors.es.net
  * extras: centos.arvixe.com
  * updates: repos.lax.quadranet.com
Resolving Dependencies
  --> Running transaction check
  --> Package tcpdump.x86_64 14:4.5.1-2.el7 will be installed
  --> Finished Dependency Resolution
tcpdump-4.5.1-2.el7.x86_64.rpm | 387 kB  00:00
Running transaction check
Running transaction test
Transaction test succeeded
Running transaction
  Installing: 14:tcpdump-4.5.1-2.el7.x86_64 1/1
  Verifying : 14:tcpdump-4.5.1-2.el7.x86_64 1/1
Installed:
tcpdump.x86_64 14:4.5.1-2.el7
Complete!

The package manager for FreeBSD is pkg.

freebsd# sudo pkg install -y tcpdump
Updating FreeBSD repository catalogue...
Fetching meta.txz: 100% 944 B 0.9kB/s 00:01
Fetching packagesite.txz: 100% 5 MiB 5.5MB/s 00:01
Processing entries: 100%
FreeBSD repository update completed. 24632 packages processed.
All repositories are up-to-date.
The following 2 package(s) will be affected (of 0 checked):

New packages to be INSTALLED:
tcpdump: 4.7.4
libsmi: 0.4.8_1

The process will require 17 MiB more space.
2 MiB to be downloaded.
Fetching tcpdump-4.7.4.txz: 100% 301 KiB 307.7kB/s 00:01
Fetching libsmi-0.4.8_1.txz: 100% 2 MiB 2.0MB/s 00:01
Checking integrity... done (0 conflicting)
[1/2] Installing libsmi-0.4.8_1...
[1/2] Extracting libsmi-0.4.8_1: 100%
[2/2] Installing tcpdump-4.7.4...
[2/2] Extracting tcpdump-4.7.4: 100%
Building software from source code

As an illustration, here’s how you build a version of tcpdump from the source code.

The first chore is to identify the code. Software maintainers sometimes keep an index of releases on the project’s web site that are downloadable as tarballs. For open source projects, you’re most likely to find the code in a Git repository.

The tcpdump source is kept on GitHub. Clone the repository in the /tmp directory, create a branch of the tagged version you want to build, then unpack, configure, build, and install it:

```
redhat$ cd /tmp
redhat$ git clone https://github.com/the-tcpdump-group/tcpdump.git
<status messages as repository is cloned>
redhat$ cd tcpdump
redhat$ git checkout tags/tcpdump-4.7.4 -b tcpdump-4.7.4
Switched to a new branch 'tcpdump-4.7.4'
redhat$ ./configure
checking build system type... x86_64-unknown-linux-gnu
checking host system type... x86_64-unknown-linux-gnu
checking for gcc... gcc
checking whether the C compiler works... yes
...
redhat$ make
<several pages of compilation output>
redhat$ sudo make install
<files are moved in to place>
```

This configure/make/make install sequence is common to most software written in C and works on all UNIX and Linux systems. It’s always a good idea to check the package’s INSTALL or README file for specifics. You must have the development environment and any package-specific prerequisites installed. (In the case of tcpdump, libpcap and its libraries are prerequisites.)

You’ll often need to tweak the build configuration, so use ./configure --help to see the options available for each particular package. Another useful configure option is --prefix=directory, which lets you compile the software for installation somewhere other than /usr/local, which is usually the default.
Installing from a web script

Cross-platform software bundles increasingly offer an expedited installation process that’s driven by a shell script you download from the web with `curl`, `fetch`, or `wget`. These are all simple HTTP clients that download the contents of a URL to a local file or, optionally, print the contents to their standard output. For example, to set up a machine as a Salt client, you can run the following commands:

```
$ curl -o /tmp/saltboot -SL https://bootstrap.saltstack.com
$ sudo sh /tmp/saltboot
```

The bootstrap script investigates the local environment, then downloads, installs, and configures an appropriate version of the software. This type of installation is particularly common in cases where the process itself is somewhat complex, but the vendor is highly motivated to make things easy for users. (Another good example is RVM; see this page.)

See Chapter 6 for more information about package installation.

This installation method is perfectly fine, but it raises a couple of issues that are worth mentioning. To begin with, it leaves no proper record of the installation for future reference. If your operating system offers a packagized version of the software, it’s usually preferable to install the package instead of running a web installer. Packages are easy to track, upgrade, and remove. (On the other hand, most OS-level packages are out of date. You probably won’t end up with the most current version of the software.)

See this page for details on HTTPS’s chain of trust.

Be very suspicious if the URL of the boot script is not secure (that is, it does not start with https:). Unsecured HTTP is trivial to hijack, and installation URLs are of particular interest to hackers because they know you’re likely to run, as root, whatever code comes back. By contrast, HTTPS validates the identity of the server through a cryptographic chain of trust. Not foolproof, but reliable enough.

A few vendors publicize an HTTP installation URL that automatically redirects to an HTTPS version. This is dumb and is in fact no more secure than straight-up HTTP. There’s nothing to prevent the initial HTTP exchange from being intercepted, so you might never reach the vendor’s redirect. However, the existence of such redirects does mean it’s worth trying your own substitution of https for http in insecure URLs. More often than not, it works just fine.

The shell accepts script text on its standard input, and this feature enables tidy, one-line installation procedures such as the following:
$ curl -L https://badvendor.com | sudo sh

However, there’s a potential issue with this construction in that the root shell still runs even if `curl` outputs a partial script and then fails—say, because of a transient network glitch. The end result is unpredictable and potentially not good.

We are not aware of any documented cases of problems attributable to this cause. Nevertheless, it is a plausible failure mode. More to the point, piping the output of `curl` to a shell has entered the collective sysadmin unconscious as a prototypical rookie blunder, so if you must do it, at least keep it on the sly.

The fix is easy: just save the script to a temporary file, then run the script in a separate step after the download successfully completes.
1.11 Where to host

Operating systems and software can be hosted in private data centers, at co-location facilities, on a cloud platform, or on some combination of these. Most burgeoning startups choose the cloud. Established enterprises are likely to have existing data centers and may run a private cloud internally.

The most practical choice, and our recommendation for new projects, is a public cloud provider. These facilities offer numerous advantages over data centers:

- No capital expenses and low initial operating costs
- No need to install, secure, and manage hardware
- On-demand adjustment of storage, bandwidth, and compute capacity
- Ready-made solutions for common ancillary needs such as databases, load balancers, queues, monitoring, and more
- Cheaper and simpler implementation of highly available/redundant systems

Early cloud systems acquired a reputation for inferior security and performance, but these are no longer major concerns. These days, most of our administration work is in the cloud. See Chapter 9 for a general introduction to this space.

Our preferred cloud platform is the leader in the space: Amazon Web Services (AWS). Gartner, a leading technology research firm, found that AWS is ten times the size of all competitors combined. AWS innovates rapidly and offers a much broader array of services than does any other provider. It also has a reputation for excellent customer service and supports a large and engaged community. AWS offers a free service tier to cut your teeth on, including a year’s use of a low powered cloud server.

Google Cloud Platform (GCP) is aggressively improving and marketing its products. Some claim that its technology is unmatched by other providers. GCP’s growth has been slow, in part due to Google’s reputation for dropping support for popular offerings. However, its customer-friendly pricing terms and unique features are appealing differentiators.

DigitalOcean is a simpler service with a stated goal of high performance. Its target market is developers, whom it woos with a clean API, low pricing, and extremely fast boot times. DigitalOcean is a strong proponent of open source software, and their tutorials and guides for popular Internet technologies are some of the best available.
1.12 Specialization and Adjacent Disciplines

System administrators do not exist in a vacuum; a team of experts is required to build and maintain a complex network. This section describes some of the roles with which system administrators overlap in skills and scope. Some administrators choose to specialize in one or more of these areas.

Your goal as a system administrator, or as a professional working in any of these related areas, is to achieve the objectives of the organization. Avoid letting politics or hierarchy interfere with progress. The best administrators solve problems and share information freely with others.
DevOps

See this page for more comments on DevOps.

DevOps is not so much a specific function as a culture or operational philosophy. It aims to improve the efficiency of building and delivering software, especially at large sites that have many interrelated services and teams. Organizations with a DevOps practice promote integration among engineering teams and may draw little or no distinction between development and operations. Experts who work in this area seek out inefficient processes and replace them with small shell scripts or large and unwieldy Chef repositories.
Site reliability engineers

Site reliability engineers value uptime and correctness above all else. Monitoring networks, deploying production software, taking pager duty, planning future expansion, and debugging outages all lie within the realm of these availability crusaders. Single points of failure are site reliability engineers’ nemeses.
Security operations engineers

Security operations engineers focus on the practical, day-to-day side of an information security program. These folks install and operate tools that search for vulnerabilities and monitor for attacks on the network. They also participate in attack simulations to gauge the effectiveness of their prevention and detection techniques.
Network administrators

Network administrators design, install, configure, and operate networks. Sites that operate data centers are most likely to employ network administrators; that’s because these facilities have a variety of physical switches, routers, firewalls, and other devices that need management. Cloud platforms also offer a variety of networking options, but these usually don’t require a dedicated administrator because most of the work is handled by the provider.
Database administrators

Database administrators (sometimes known as DBAs) are experts at installing and managing database software. They manage database schemas, perform installations and upgrades, configure clustering, tune settings for optimal performance, and help users formulate efficient queries. DBAs are usually wizards with one or more query languages and have experience with both relational and nonrelational (NoSQL) databases.
Network operations center (NOC) engineers

NOC engineers monitor the real-time health of large sites and track incidents and outages. They troubleshoot tickets from users, perform routine upgrades, and coordinate actions among other teams. They can most often be found watching a wall of monitors that show graphs and measurements.
Data center technicians

Data center technicians work with hardware. They receive new equipment, track equipment inventory and life cycles, install servers in racks, run cabling, maintain power and air conditioning, and handle the daily operations of a data center. As a system administrator, it’s in your best interest to befriend data center technicians and bribe them with coffee, caffeinated soft drinks, and alcoholic beverages.
Architects

Systems architects have deep expertise in more than one area. They use their experience to design distributed systems. Their job descriptions may include defining security zones and segmentation, eliminating single points of failure, planning for future growth, ensuring connectivity among multiple networks and third parties, and other site-wide decision making. Good architects are technically proficient and generally prefer to implement and test their own designs.
1.13 RECOMMENDED READING


SALUS, PETER H. *The Daemon, the GNU & the Penguin: How Free and Open Software is Changing the World*. Reed Media Services, 2008. This fascinating history of the open source movement by UNIX’s best-known historian is also available at groklaw.com under the Creative Commons license. The URL for the book itself is quite long; look for a current link at groklaw.com or try this compressed equivalent: tinyurl.com/d6u7j.

System administration and DevOps


Limoncelli, Thomas A., Christina J. Hogan, and Strata R. Chalup. The Practice of System and Network Administration (2nd Edition). Reading, MA: Addison-Wesley, 2008. This is a good book with particularly strong coverage of the policy and procedural aspects of system administration. The authors maintain a system administration blog at everythingsysadmin.com.

Limoncelli, Thomas A., Christina J. Hogan, and Strata R. Chalup. The Practice of Cloud System Administration. Reading, MA: Addison-Wesley, 2014. From the same authors as the previous title, now with a focus on distributed systems and cloud computing.
Essential tools


Booting and System Management Daemons

“Booting” is the standard term for “starting up a computer.” It’s a shortened form of the word “bootstrapping,” which derives from the notion that the computer has to “pull itself up by its own bootstraps.”

The boot process consists of a few broadly defined tasks:

- Finding, loading, and running bootstrapping code
- Finding, loading, and running the OS kernel
- Running startup scripts and system daemons
- Maintaining process hygiene and managing system state transitions

The activities included in that last bullet point continue as long as the system remains up, so the line between bootstrapping and normal operation is inherently a bit blurry.
2.1 **Boot process overview**

Startup procedures have changed a lot in recent years. The advent of modern (UEFI) BIOSs has simplified the early stages of booting, at least from a conceptual standpoint. In later stages, most Linux distributions now use a system manager daemon called **systemd** instead of the traditional UNIX **init**. **systemd** streamlines the boot process by adding dependency management, support for concurrent startup processes, and a comprehensive approach to logging, among other features.

Boot management has also changed as systems have migrated into the cloud. The drift toward virtualization, cloud instances, and containerization has reduced the need for administrators to touch physical hardware. Instead, we now have image management, APIs, and control panels.

During bootstrapping, the kernel is loaded into memory and begins to execute. A variety of initialization tasks are performed, and the system is then made available to users. The general overview of this process is shown in Exhibit A.

**Exhibit A: Linux & UNIX boot process**

![Exhibit A: Linux & UNIX boot process diagram](image)

Administrators have little direct, interactive control over most of the steps required to boot a system. Instead, admins can modify bootstrap configurations by editing config files for the system startup scripts or by changing the arguments the boot loader passes to the kernel.

Before the system is fully booted, filesystems must be checked and mounted and system daemons started. These procedures are managed by a series of shell scripts (sometimes called
“init scripts” or unit files that are run in sequence by init or parsed by systemd. The exact layout of the startup scripts and the manner in which they are executed varies among systems. We cover the details later in this chapter.
2.2 SYSTEM FIRMWARE

When a machine is powered on, the CPU is hardwired to execute boot code stored in ROM. On virtualized systems, this “ROM” may be imaginary, but the concept remains the same.

The system firmware typically knows about all the devices that live on the motherboard, such as SATA controllers, network interfaces, USB controllers, and sensors for power and temperature. (Virtual systems pretend to have this same set of devices.) In addition to allowing hardware-level configuration of these devices, the firmware lets you either expose them to the operating system or disable and hide them.

On physical (as opposed to virtualized) hardware, most firmware offers a user interface. However, it’s generally crude and a bit tricky to access. You need control of the computer and console, and must press a particular key immediately after powering on the system. Unfortunately, the identity of the magic key varies by manufacturer; see if you can glimpse a cryptic line of instructions at the instant the system first powers on. (You might find it helpful to disable the monitor’s power management features temporarily.) Barring that, try Delete, Control, F6, F8, F10, or F11. For the best chance of success, tap the key several times, then hold it down.

During normal bootstrapping, the system firmware probes for hardware and disks, runs a simple set of health checks, and then looks for the next stage of bootstrapping code. The firmware UI lets you designate a boot device, usually by prioritizing a list of available options (e.g., “try to boot from the DVD drive, then a USB drive, then a hard disk”).

In most cases, the system’s disk drives populate a secondary priority list. To boot from a particular drive, you must both set it as the highest-priority disk and make sure that “hard disk” is enabled as a boot medium.
BIOS vs. UEFI

Traditional PC firmware was called the BIOS, for Basic Input/Output System. Over the last decade, however, BIOS has been supplanted by a more formalized and modern standard, the Unified Extensible Firmware Interface (UEFI). You’ll often see UEFI referred to as “UEFI BIOS,” but for clarity, we’ll reserve the term BIOS for the legacy standard in this chapter. Most systems that implement UEFI can fall back to a legacy BIOS implementation if the operating system they’re booting doesn’t support UEFI.

UEFI is the current revision of an earlier standard, EFI. References to the name EFI persist in some older documentation and even in some standard terms, such as “EFI system partition.” In all but the most technically explicit situations, you can treat these terms as equivalent.

UEFI support is pretty much universal on new PC hardware these days, but plenty of BIOS systems remain in the field. Moreover, virtualized environments often adopt BIOS as their underlying boot mechanism, so the BIOS world isn’t in danger of extinction just yet.

As much as we’d prefer to ignore BIOS and just talk about UEFI, it’s likely that you’ll encounter both types of systems for years to come. UEFI also builds-in several accommodations to the old BIOS regime, so a working knowledge of BIOS can be quite helpful for deciphering the UEFI documentation.
Legacy BIOS

Partitioning is a way to subdivide physical disks. See this page for a more detailed discussion.

Traditional BIOS assumes that the boot device starts with a record called the MBR (Master Boot Record). The MBR includes both a first-stage boot loader (aka “boot block”) and a primitive disk partitioning table. The amount of space available for the boot loader is so small (less than 512 bytes) that it’s not able to do much other than load and run a second-stage boot loader.

Neither the boot block nor the BIOS is sophisticated enough to read any type of standard filesystem, so the second-stage boot loader must be kept somewhere easy to find. In one typical scenario, the boot block reads the partitioning information from the MBR and identifies the disk partition marked as “active.” It then reads and executes the second-stage boot loader from the beginning of that partition. This scheme is known as a volume boot record.

Alternatively, the second-stage boot loader can live in the dead zone that lies between the MBR and the beginning of the first disk partition. For historical reasons, the first partition doesn’t start until the 64th disk block, so this zone normally contains at least 32KB of storage: still not a lot, but enough to store a filesystem driver. This storage scheme is commonly used by the GRUB boot loader; see GRUB: the GRand Unified Boot loader.

To effect a successful boot, all components of the boot chain must be properly installed and compatible with one another. The MBR boot block is OS-agnostic, but because it assumes a particular location for the second stage, there may be multiple versions that can be installed. The second-stage loader is generally knowledgeable about operating systems and filesystems (it may support several of each), and usually has configuration options of its own.
The UEFI specification includes a modern disk partitioning scheme known as GPT (GUID Partition Table, where GUID stands for “globally unique identifier”). UEFI also understands FAT (File Allocation Table) filesystems, a simple but functional layout that originated in MS-DOS. These features combine to define the concept of an EFI System Partition (ESP). At boot time, the firmware consults the GPT partition table to identify the ESP. It then reads the configured target application directly from a file in the ESP and executes it.

Because the ESP is just a generic FAT filesystem, it can be mounted, read, written, and maintained by any operating system. No “mystery meat” boot blocks are required anywhere on the disk. (Truth be told, UEFI does maintain an MBR-compatible record at the beginning of each disk to facilitate interoperability with BIOS systems. BIOS systems can’t see the full GPT-style partition table, but they at least recognize the disk as having been formatted. Be careful not to run MBR-specific administrative tools on GPT disks. They may think they understand the disk layout, but they do not.)

In the UEFI system, no boot loader at all is technically required. The UEFI boot target can be a UNIX or Linux kernel that has been configured for direct UEFI loading, thus effecting a loader-less bootstrap. In practice, though, most systems still use a boot loader, partly because that makes it easier to maintain compatibility with legacy BIOSes.

UEFI saves the pathname to load from the ESP as a configuration parameter. With no configuration, it looks for a standard path, usually `/efi/boot/bootx64.efi` on modern Intel systems. A more typical path on a configured system (this one for Ubuntu and the GRUB boot loader) would be `/efi/ubuntu/grubx64.efi`. Other distributions follow a similar convention.

UEFI defines standard APIs for accessing the system’s hardware. In this respect, it’s something of a miniature operating system in its own right. It even provides for UEFI-level add-on device drivers, which are written in a processor-independent language and stored in the ESP. Operating systems can use the UEFI interface, or they can take over direct control of the hardware.

Because UEFI has a formal API, you can examine and modify UEFI variables (including boot menu entries) on a running system. For example, `efibootmgr -v` shows the following summary of the boot configuration:
efibootmgr lets you change the boot order, select the next configured boot option, or even create and destroy boot entries. For example, to set the boot order to try the system drive before trying the network, and to ignore other boot options, we could use the command

```bash
$ efibootmgr -v
BootCurrent: 0004
BootOrder: 0000,0001,0002,0004,0003
Boot0000* EFI DVD/CDROM PciRoot(0x0)/Pci(0x1f,0x2)/Sata(1,0,0)
Boot0001* EFI Hard Drive PciRoot(0x0)/Pci(0x1f,0x2)/Sata(0,0,0)
Boot0002* EFI Network PciRoot(0x0)/Pci(0x5,0x0)/MAC(001c42fb5baf,0)
Boot0003* EFI Internal Shell MemoryMapped(11,0x7ed5d000,0x7f0dcfff)/FvFile(c57ad6b7-0515-40a8-9d21-551652854e37)
Boot0004* ubuntu HD(1,GPT,020c8d3e-fd8c-4880-9b61-ef4cfc3d76c,0x800,0x100000)/File(EFI/ubuntu/shimx64.efi)
```

The ability to modify the UEFI configuration from user space means that the firmware’s configuration information is mounted read/write—a blessing and a curse. On systems (typically, those with systemd) that allow write access by default, `rm -rf /` can be enough to permanently destroy the system at the firmware level; in addition to removing files, `rm` also removes variables and other UEFI information accessible through /sys. Yikes! Don’t try this at home! (See goo.gl/QMSiSG, a link to a Phoronix article, for some additional details.)
2.3 **Boot loaders**

Most bootstrapping procedures include the execution of a boot loader that is distinct from both the BIOS/UEFI code and the OS kernel. It’s also separate from the initial boot block on a BIOS system, if you’re counting steps.

The boot loader’s main job is to identify and load an appropriate operating system kernel. Most boot loaders can also present a boot-time user interface that lets you select which of several possible kernels or operating systems to invoke.

Another task that falls to the boot loader is the marshaling of configuration arguments for the kernel. The kernel doesn’t have a command line per se, but its startup option handling will seem eerily similar from the shell. For example, the argument `single` or `-s` usually tells the kernel to enter single-user mode instead of completing the normal boot process.

Such options can be hard-wired into the boot loader’s configuration if you want them used on every boot, or they can be provided on the fly through the boot loader’s UI.

In the next few sections, we discuss GRUB (the Linux world’s predominant boot loader) and the boot loaders used with FreeBSD.
2.4 GRUB: the GRand Unified Boot loader

GRUB, developed by the GNU Project, is the default boot loader on most Linux distributions. The GRUB lineage has two main branches: the original GRUB, now called GRUB Legacy, and the newer, extra-crispy GRUB 2, which is the current standard. Make sure you know which GRUB you’re dealing with, as the two versions are quite different.

GRUB 2 has been the default boot manager for Ubuntu since version 9.10, and it recently became the default for RHEL 7. All our example Linux distributions use it as their default. In this book we discuss only GRUB 2, and we refer to it simply as GRUB.

FreeBSD has its own boot loader (covered in more detail in *The FreeBSD boot process*). However, GRUB is perfectly happy to boot FreeBSD, too. This might be an advantageous configuration if you’re planning to boot multiple operating systems on a single computer. Otherwise, the FreeBSD boot loader is more than adequate.
GRUB lets you specify parameters such as the kernel to boot (specified as a GRUB “menu entry”) and the operating mode to boot into.

Since this configuration information is needed at boot time, you might imagine that it would be stored somewhere strange, such as the system’s NVRAM or the disk blocks reserved for the boot loader. In fact, GRUB understands most of the filesystems in common use and can usually find its way to the root filesystem on its own. This feat lets GRUB read its configuration from a regular text file.

The config file is called `grub.cfg`, and it’s usually kept in `/boot/grub` (/`boot/grub2` in Red Hat and CentOS) along with a selection of other resources and code modules that GRUB might need to access at boot time. Changing the boot configuration is a simple matter of updating the `grub.cfg` file.

Although you can create the `grub.cfg` file yourself, it’s more common to generate it with the `grub-mkconfig` utility, which is called `grub2-mkconfig` on Red Hat and CentOS and wrapped as `update-grub` on Debian and Ubuntu. In fact, most distributions assume that `grub.cfg` can be regenerated at will, and they do so automatically after updates. If you don’t take steps to prevent this, your handcrafted `grub.cfg` file will get clobbered.

As with all things Linux, distributions configure `grub-mkconfig` in a variety of ways. Most commonly, the configuration is specified in `/etc/default/grub` in the form of `sh` variable assignments. Table 2.1 shows some of the commonly modified options.

### Table 2.1: Common GRUB configuration options from `/etc/default/grub`

<table>
<thead>
<tr>
<th>Shell variable name</th>
<th>Contents or function</th>
</tr>
</thead>
<tbody>
<tr>
<td>GRUB_BACKGROUND</td>
<td>Background image&lt;sup&gt;a&lt;/sup&gt;</td>
</tr>
<tr>
<td>GRUB_CMDLINE_LINUX</td>
<td>Kernel parameters to add to menu entries for Linux&lt;sup&gt;b&lt;/sup&gt;</td>
</tr>
<tr>
<td>GRUB_DEFAULT</td>
<td>Number or title of the default menu entry</td>
</tr>
<tr>
<td>GRUB_DISABLE_RECOVERY</td>
<td>Prevents the generation of recovery mode entries</td>
</tr>
<tr>
<td>GRUB_PRELOAD_MODULES</td>
<td>List of GRUB modules to be loaded as early as possible</td>
</tr>
<tr>
<td>GRUB_TIMEOUT</td>
<td>Seconds to display the boot menu before autoboot</td>
</tr>
</tbody>
</table>

<sup>a.</sup> The background image must be a `.png`, `.tga`, `.jpg`, or `.jpeg` file.

<sup>b.</sup> Table 2.3 lists some of the available options.

After editing `/etc/default/grub`, run `update-grub` or `grub2-mkconfig` to translate your configuration into a proper `grub.cfg` file. As part of the configuration-building process, these
commands inventory the system’s bootable kernels, so they can be useful to run after you make kernel changes even if you haven’t explicitly changed the GRUB configuration.

You may need to edit the /etc/grub.d/40_custom file to change the order in which kernels are listed in the boot menu (after you create a custom kernel, for example), set a boot password, or change the names of boot menu items. As usual, run update-grub or grub2-mkconfig after making changes.

As an example, here’s a 40_custom file that invokes a custom kernel on an Ubuntu system:

```bash
#!/bin/sh
exec tail -n +3 $0

# This file provides an easy way to add custom menu entries. Just type
# the menu entries you want to add after this comment. Be careful not to
# change the 'exec tail' line above.

menuentry 'My Awesome Kernel' {
    set root='(hd0,msdos1)'
    linux  /awesome_kernel root=UUID=XXX-XXX-XXX ro quiet
    initrd  /initrd.img-awesome_kernel
}
```

See [this page](#) for more information about mounting filesystems.

In this example, GRUB loads the kernel from /awesome_kernel. Kernel paths are relative to the boot partition, which historically was mounted as /boot but with the advent of UEFI now is likely an unmounted EFI System Partition. Use gpart show and mount to examine your disk and determine the state of the boot partition.

The GRUB command line

GRUB supports a command-line interface for editing config file entries on the fly at boot time. To enter command-line mode, type `c` at the GRUB boot screen.

From the command line, you can boot operating systems that aren’t listed in the `grub.cfg` file, display system information, and perform rudimentary filesystem testing. Anything that can be done through `grub.cfg` can also be done through the command line.

Press the `<Tab>` key to see a list of possible commands. Table 2.2 shows some of the more useful ones.

Table 2.2: GRUB commands

<table>
<thead>
<tr>
<th>Cmd</th>
<th>Function</th>
</tr>
</thead>
<tbody>
<tr>
<td>boot</td>
<td>Boots the system from the specified kernel image</td>
</tr>
<tr>
<td>help</td>
<td>Gets interactive help for a command</td>
</tr>
<tr>
<td>linux</td>
<td>Loads a Linux kernel</td>
</tr>
<tr>
<td>reboot</td>
<td>Reboots the system</td>
</tr>
<tr>
<td>search</td>
<td>Searches devices by file, filesystem label, or UUID</td>
</tr>
<tr>
<td>usb</td>
<td>Tests USB support</td>
</tr>
</tbody>
</table>

Linux kernel options

See *Chapter 11* for more about kernel parameters.

Kernel startup options typically modify the values of kernel parameters, instruct the kernel to probe for particular devices, specify the path to the **init** or **systemd** process, or designate a particular root device. Table 2.3 shows a few examples.

<table>
<thead>
<tr>
<th>Option</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>debug</td>
<td>Turns on kernel debugging</td>
</tr>
<tr>
<td>init=/bin/bash</td>
<td>Starts only the <strong>bash</strong> shell; useful for emergency recovery</td>
</tr>
<tr>
<td>root=/dev/foo</td>
<td>Tells the kernel to use /dev/foo as the root device</td>
</tr>
<tr>
<td>single</td>
<td>Boots to single-user mode</td>
</tr>
</tbody>
</table>

When specified at boot time, kernel options are not persistent. Edit the appropriate kernel line in `/etc/grub.d/40_custom` or `/etc/defaults/grub` (the variable named `GRUB_CMDLINE_LINUX`) to make the change permanent across reboots.

Security patches, bug fixes, and features are all regularly added to the Linux kernel. Unlike other software packages, however, new kernel releases typically do not replace old ones. Instead, the new kernels are installed side by side with the previous versions so that you can return to an older kernel in the event of problems.

This convention helps administrators back out of an upgrade if a kernel patch breaks their system, although it also means that the boot menu tends to get cluttered with old versions of the kernel. Try choosing a different kernel if your system won’t boot after an update.
2.5 **THE FreeBSD boot process**

FreeBSD’s boot system is a lot like GRUB in that the final-stage boot loader (called **loader**) uses a filesystem-based configuration file, supports menus, and offers an interactive, command-line-like mode. **loader** is the final common pathway for both the BIOS and UEFI boot paths.
The BIOS path: boot0

As with GRUB, the full loader environment is too large to fit in an MBR boot block, so a chain of progressively more sophisticated preliminary boot loaders get loader up and running on a BIOS system.

GRUB bundles all of these components under the umbrella name “GRUB,” but in FreeBSD, the early boot loaders are part of a separate system called boot0 that’s used only on BIOS systems. boot0 has options of its own, mostly because it stores later stages of the boot chain in a volume boot record (see Legacy BIOS) rather than in front of the first disk partition.

For that reason, the MBR boot record needs a pointer to the partition it should use to continue the boot process. Normally, all this is automatically set up for you as part of the FreeBSD installation process, but if you should ever need to adjust the configuration, you can do so with the boot0cfg command.
The UEFI path

On UEFI systems, FreeBSD creates an EFI system partition and installs boot code there under the path /boot/bootx64.efi. This is the default path that UEFI systems check at boot time (at least on modern PC platforms), so no firmware-level configuration should be needed other than ensuring that device boot priorities are properly set.

Don’t confuse the /boot directory in the EFI system partition with the /boot directory in the FreeBSD root filesystem. They are separate and serve different purposes, although of course both are bootstrapping related.

By default, FreeBSD doesn’t keep the EFI system partition mounted after booting. You can inspect the partition table with gpart to identify it:

```bash
$ gpart show
=>
  40 134217648   ada0  GPT (64G)
  40   1600   1  efi (800K)
 1640 127924664   2  freebsd-ufs (61G)
 127926304   6291383   3  freebsd-swap (3.0G)
 134217687   1   - free - (512B)
```

See this page for more information about mounting filesystems.

Although you can mount the ESP if you’re curious to see what’s in it (use mount’s -t msdos option), the whole filesystem is actually just a copy of an image found in /boot/boot1.efifat on the root disk. No user-serviceable parts inside.

If the ESP partition gets damaged or removed, you can re-create it by setting up the partition with gpart and then copying in the filesystem image with dd:

```bash
$ sudo dd if=/boot/boot1.efifat of=/dev/ada0p1
```

Once the first-stage UEFI boot loader finds a partition of type freebsd-ufs, it loads a UEFI version of the loader software from /boot/loader.efi. From there, booting proceeds as under BIOS, with loader determining the kernel to load, the kernel parameters to set, and so on. (As of FreeBSD 10.1, it is possible to use ZFS as the root partition on a UEFI system as well.)
loader configuration

loader is actually a scripting environment, and the scripting language is Forth. This is a remarkable and interesting fact if you’re a historian of programming languages, and unimportant otherwise. There’s a bunch of Forth code stored under /boot that controls loader’s operations, but it’s designed to be self-contained—you needn’t learn Forth.

The Forth scripts execute /boot/loader.conf to obtain the values of configuration variables, so that’s where your customizations should go. Don’t confuse this file with /boot/defaults/loader.conf, which contains the configuration defaults and isn’t intended for modification. Fortunately, the variable assignments in loader.conf are syntactically similar to standard sh assignments.

The man pages for loader and loader.conf give the dirt on all the boot loader options and the configuration variables that control them. Some of the more interesting options include those for protecting the boot menu with a password, changing the splash screen displayed at boot, and passing kernel options.
**loader commands**

*loader* understands a variety of interactive commands. For example, to locate and boot an alternate kernel, you’d use a sequence of commands like this:

Type ‘?’ for a list of commands, ‘help’ for more detailed help.

```plaintext
OK ls
/
  d .snap
  d dev
...
d rescue
1 home
...
OK unload
OK load /boot/kernel/kernel.old
/boot/kernel/kernel.old text=0xf8f898 data=0x124 ... b077]
OK boot
```

Here, we listed the contents of the (default) root filesystem, unloaded the default kernel (/boot/kernel/kernel), loaded an older kernel (/boot/kernel/kernel.old), and then continued the boot process.

See *man loader* for complete documentation of the available commands.
2.6 SYSTEM MANAGEMENT DAEMONS

Once the kernel has been loaded and has completed its initialization process, it creates a complement of “spontaneous” processes in user space. They’re called spontaneous processes because the kernel starts them autonomously—in the normal course of events, new processes are created only at the behest of existing processes.

Most of the spontaneous processes are really part of the kernel implementation. They don’t necessarily correspond to programs in the filesystem. They’re not configurable, and they don’t require administrative attention. You can recognize them in `ps` listings (see this page) by their low PIDs and by the brackets around their names (for example, `[pagedaemon]` on FreeBSD or `[kdump]` on Linux).

The exception to this pattern is the system management daemon. It has process ID 1 and usually runs under the name `init`. The system gives `init` a couple of special privileges, but for the most part it’s just a user-level program like any other daemon.
Responsibilities of init

init has multiple functions, but its overarching goal is to make sure the system runs the right complement of services and daemons at any given time.

To serve this goal, init maintains a notion of the mode in which the system should be operating. Some commonly defined modes:

- Single-user mode, in which only a minimal set of filesystems is mounted, no services are running, and a root shell is started on the console
- Multiuser mode, in which all customary filesystems are mounted and all configured network services have been started, along with a window system and graphical login manager for the console
- Server mode, similar to multiuser mode, but with no GUI running on the console

Don’t take these mode names or descriptions too literally; they’re just examples of common operating modes that most systems define in one way or another.

Every mode is associated with a defined complement of system services, and the initialization daemon starts or stops services as needed to bring the system’s actual state into line with the currently active mode. Modes can also have associated milepost tasks that run whenever the mode begins or ends.

As an example, init normally takes care of many different startup chores as a side effect of its transition from bootstrapping to multiuser mode. These may include

- Setting the name of the computer
- Setting the time zone
- Checking disks with fsck
- Mounting filesystems
- Removing old files from the /tmp directory
- Configuring network interfaces
- Configuring packet filters
- Starting up other daemons and network services

init has very little built-in knowledge about these tasks. In simply runs a set of commands or scripts that have been designated for execution in that particular context.
Implementations of init

Today, three very different flavors of system management processes are in widespread use:

- An init styled after the init from AT&T’s System V UNIX, which we refer to as “traditional init.” This was the predominant init used on Linux systems until the debut of systemd.

- An init variant that derives from BSD UNIX and is used on most BSD-based systems, including FreeBSD, OpenBSD, and NetBSD. This one is just as tried-and-true as the SysV init and has just as much claim to being called “traditional,” but for clarity we refer to it as “BSD init.” This variant is quite simple in comparison with SysV-style init. We discuss it separately starting on this page.

- A more recent contender called systemd which aims to be one-stop-shopping for all daemon- and state-related issues. As a consequence, systemd carves out a significantly larger territory than any historical version of init. That makes it somewhat controversial, as we discuss below. Nevertheless, all our example Linux distributions have now adopted systemd.

Although these implementations are the predominant ones today, they’re far from being the only choices. Apple’s macOS, for example, uses a system called launchd. Until it adopted systemd, Ubuntu used another modern init variant called Upstart.

On Linux systems, you can theoretically replace your system’s default init with whichever variant you prefer. But in practice, init is so fundamental to the operation of the system that a lot of add-on software is likely to break. If you can’t abide systemd, standardize on a distribution that doesn’t use it.
**Traditional init**

In the traditional **init** world, system modes (e.g., single-user or multiuser) are known as “run levels.” Most run levels are denoted by a single letter or digit.

Traditional **init** has been around since the early 80s, and grizzled folks in the anti-**systemd** camp often cite the principle, “If it ain’t broke, don’t fix it.” That said, traditional **init** does have a number of notable shortcomings.

To begin with, the traditional **init** on its own is not really powerful enough to handle the needs of a modern system. Most systems that use it actually have a standard and fixed **init** configuration that never changes. That configuration points to a second tier of shell scripts that do the actual work of changing run levels and letting administrators make configuration changes.

The second layer of scripts maintains yet a third layer of daemon- and system-specific scripts, which are cross-linked to run-level-specific directories that indicate what services are supposed to be running at what run level. It’s all a bit hackish and unsightly.

More concretely, this system has no general model of dependencies among services, so it requires that all startups and takedowns be run in a numeric order that’s maintained by the administrator. Later actions can’t run until everything ahead of them has finished, so it’s impossible to execute actions in parallel, and the system takes a long time to change states.
systemd vs. the world

Few issues in the Linux space have been more hotly debated than the migration from traditional `init` to `systemd`. For the most part, complaints center on `systemd`’s seemingly ever-increasing scope.

`systemd` takes all the `init` features formerly implemented with sticky tape, shell script hacks, and the sweat of administrators and formalizes them into a unified field theory of how services should be configured, accessed, and managed.

See Chapter 6, Software Installation and Management, for more information about package management.

Much like a package management system, `systemd` defines a robust dependency model, not only among services but also among “targets,” `systemd`’s term for the operational modes that traditional `init` calls run levels. `systemd` not only manages processes in parallel, but also manages network connections (`networkd`), kernel log entries (`journald`), and logins (`logind`).

The anti-`systemd` camp argues that the UNIX philosophy is to keep system components small, simple, and modular. A component as fundamental as `init`, they say, should not have monolithic control over so many of the OS’s other subsystems. `systemd` not only breeds complexity, but also introduces potential security weaknesses and muddies the distinction between the OS platform and the services that run on top of it.

`systemd` has also received criticism for imposing new standards and responsibilities on the Linux kernel, for its code quality, for the purported unresponsiveness of its developers to bug reports, for the functional design of its basic features, and for looking at people funny. We can’t fairly address these issues here, but you may find it informative to peruse the Arguments against `systemd` section at without-systemd.org, the Internet’s premier `systemd` hate site.
The architectural objections to **systemd** outlined above are all reasonable points. **systemd** does indeed display most of the telltale stigmata of an overengineered software project.

In practice, however, many administrators quite like **systemd**, and we fall squarely into this camp. Ignore the controversy for a bit and give **systemd** a chance to win your love. Once you’ve become accustomed to it, you will likely find yourself appreciating its many merits.

At the very least, keep in mind that the traditional **init** that **systemd** displaces was no national treasure. If nothing else, **systemd** delivers some value just by eliminating a few of the unnecessary differences among Linux distributions.

The debate really doesn’t matter anymore because the **systemd** coup is over. The argument was effectively settled when Red Hat, Debian, and Ubuntu switched. Many other Linux distributions are now adopting **systemd**, either by choice or by being dragged along, kicking and screaming, by their upstream distributions.

Traditional **init** still has a role to play when a distribution either targets a small installation footprint or doesn’t need **systemd**’s advanced process management functions. There’s also a sizable population of revanchists who disdain **systemd** on principle, so some Linux distributions are sure to keep traditional **init** alive indefinitely as a form of protest theater.

Nevertheless, we don’t think that traditional **init** has enough of a future to merit a detailed discussion in this book. For Linux, we mostly limit ourselves to **systemd**. We also discuss the mercifully simple system used by FreeBSD, starting on [this page](#).
2.7 SYSTEMD IN DETAIL

The configuration and control of system services is an area in which Linux distributions have traditionally differed the most from one another. systemd aims to standardize this aspect of system administration, and to do so, it reaches further into the normal operations of the system than any previous alternative.

systemd is not a single daemon but a collection of programs, daemons, libraries, technologies, and kernel components. A post on the systemd blog at Opointer.de/blog notes that a full build of the project generates 69 different binaries. Think of it as a scrumptious buffet at which you are forced to consume everything.

Since systemd depends heavily on features of the Linux kernel, it’s a Linux-only proposition. You won’t see it ported to BSD or to any other variant of UNIX within the next five years.
Units and unit files

An entity that is managed by systemd is known generically as a unit. More specifically, a unit can be “a service, a socket, a device, a mount point, an automount point, a swap file or partition, a startup target, a watched filesystem path, a timer controlled and supervised by systemd, a resource management slice, a group of externally created processes, or a wormhole into an alternate universe.” OK, we made up the part about the alternate universe (the rest is from the systemd.unit man page), but that still covers a lot of territory.

Within systemd, the behavior of each unit is defined and configured by a unit file. In the case of a service, for example, the unit file specifies the location of the executable file for the daemon, tells systemd how to start and stop the service, and identifies any other units that the service depends on.

We explore the syntax of unit files in more detail soon, but here’s a simple example from an Ubuntu system as an appetizer. This unit file is rsync.service; it handles startup of the rsync daemon that mirrors filesystems.

See this page for more information about rsync.

[Unit]
Description=fast remote file copy program daemon
ConditionPathExists=/etc/rsyncd.conf

[Service]
ExecStart=/usr/bin/rsync --daemon --no-detach

[Install]
WantedBy=multi-user.target

If you recognize this as the file format used by MS-DOS .ini files, you are well on your way to understanding both systemd and the anguish of the systemd haters.

Unit files can live in several different places. /usr/lib/systemd/system is the main place where packages deposit their unit files during installation; on some systems, the path is /lib/systemd/system instead. The contents of this directory are considered stock, so you shouldn’t modify them. Your local unit files and customizations can go in /etc/systemd/system. There’s also a unit directory in /run/systemd/system that’s a scratch area for transient units.

systemd maintains a telescopic view of all these directories, so they’re pretty much equivalent. If there’s any conflict, the files in /etc have the highest priority.

By convention, unit files are named with a suffix that varies according to the type of unit being configured. For example, service units have a .service suffix and timers use .timer. Within the
unit file, some sections (e.g., [Unit]) apply generically to all kinds of units, but others (e.g., [Service]) can appear only in the context of a particular unit type.
systemctl: manage systemd

`systemctl` is an all-purpose command for investigating the status of `systemd` and making changes to its configuration. As with Git and several other complex software suites, `systemctl`’s first argument is typically a subcommand that sets the general agenda, and subsequent arguments are specific to that particular subcommand. The subcommands could be top-level commands in their own right, but for consistency and clarity, they’re bundled into the `systemctl` omnibus.

See this page for more information about Git.

Running `systemctl` without any arguments invokes the default `list-units` subcommand, which shows all loaded and active services, sockets, targets, mounts, and devices. To show only loaded and active services, use the `--type=service` qualifier:

```
$ systemctl list-units --type=service
UNIT LOAD ACTIVE SUB DESCRIPTION
accounts-daemon.service loaded active running Accounts Service
...
wpa_supplicant.service loaded active running WPA supplicant
```

It’s also sometimes helpful to see all the installed unit files, regardless of whether or not they’re active:

```
$ systemctl list-unit-files --type=service
UNIT FILE   STATE
...
cron.service enabled
cryptdisks-early.service masked
cryptdisks.service masked
cups-browsed.service enabled
cups.service disabled
...
wpa_supplicant.service disabled
x11-common.service masked
```

188 unit files listed.

For subcommands that act on a particular unit (e.g., `systemctl status`) `systemctl` can usually accept a unit name without a unit-type suffix (e.g., `cups` instead of `cups.service`). However, the default unit type with which simple names are fleshed out varies by subcommand.

Table 2.4 shows the most common and useful `systemctl` subcommands. See the `systemctl` man page for a complete list.

Table 2.4: Commonly used systemctl subcommands
<table>
<thead>
<tr>
<th>Subcommand</th>
<th>Function</th>
</tr>
</thead>
<tbody>
<tr>
<td>list-unit-files</td>
<td>Shows installed units; optionally matching pattern</td>
</tr>
<tr>
<td>enable unit</td>
<td>Enables unit to activate at boot</td>
</tr>
<tr>
<td>disable unit</td>
<td>Prevents unit from activating at boot</td>
</tr>
<tr>
<td>isolate target</td>
<td>Changes operating mode to target</td>
</tr>
<tr>
<td>start unit</td>
<td>Activates unit immediately</td>
</tr>
<tr>
<td>stop unit</td>
<td>Deactivates unit immediately</td>
</tr>
<tr>
<td>restart unit</td>
<td>Restarts (or starts, if not running) unit immediately</td>
</tr>
<tr>
<td>status unit</td>
<td>Shows unit’s status and recent log entries</td>
</tr>
<tr>
<td>kill pattern</td>
<td>Sends a signal to units matching pattern</td>
</tr>
<tr>
<td>reboot</td>
<td>Reboots the computer</td>
</tr>
<tr>
<td>daemon-reload</td>
<td>Reloads unit files and <strong>systemd</strong> configuration</td>
</tr>
</tbody>
</table>
Unit statuses

In the output of systemctl list-unit-files above, we can see that cups.service is disabled. We can use systemctl status to find out more details:

```
$ sudo systemctl status -l cups
```

cups.service - CUPS Scheduler
   Loaded: loaded (/lib/systemd/system/cups.service; disabled; vendor
          preset: enabled)
   Active: inactive (dead) since Sat 2016-12-12 00:51:40 MST; 4s ago
          Docs: man:cupsd(8)
   Main PID: 10081 (code=exited, status=0/SUCCESS)

Dec 12 00:44:39 ulsaah systemd[1]: Started CUPS Scheduler.
Dec 12 00:44:45 ulsaah systemd[1]: Started CUPS Scheduler.
Dec 12 00:51:40 ulsaah systemd[1]: Stopping CUPS Scheduler...
Dec 12 00:51:40 ulsaah systemd[1]: Stopped CUPS Scheduler.

Here, systemctl shows us that the service is currently inactive (dead) and tells us when the process died. (Just a few seconds ago; we disabled it for this example.) It also shows (in the section marked Loaded) that the service defaults to being enabled at startup, but that it is presently disabled.

The last four lines are recent log entries. By default, the log entries are condensed so that each entry takes only one line. This compression often makes entries unreadable, so we included the -l option to request full entries. It makes no difference in this case, but it’s a useful habit to acquire.

Table 2.5 shows the statuses you’ll encounter most frequently when checking up on units.

<table>
<thead>
<tr>
<th>State</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>bad</td>
<td>Some kind of problem within systemd; usually a bad unit file</td>
</tr>
<tr>
<td>disabled</td>
<td>Present, but not configured to start autonomously</td>
</tr>
<tr>
<td>enabled</td>
<td>Installed and runnable; will start autonomously</td>
</tr>
<tr>
<td>indirect</td>
<td>Disabled, but has peers in Also clauses that may be enabled</td>
</tr>
<tr>
<td>linked</td>
<td>Unit file available through a symlink</td>
</tr>
<tr>
<td>masked</td>
<td>Banished from the systemd world from a logical perspective</td>
</tr>
<tr>
<td>static</td>
<td>Depended upon by another unit; has no install requirements</td>
</tr>
</tbody>
</table>

The enabled and disabled states apply only to unit files that live in one of systemctl’s system directories (that is, they are not linked in by a symbolic link) and that have an [Install] section in their unit files. “Enabled” units should perhaps really be thought of as “installed,” meaning that the directives in the [Install] section have been executed and that the unit is wired up to its normal
activation triggers. In most cases, this state causes the unit to be activated automatically once the system is bootstrapped.

Likewise, the disabled state is something of a misnomer because the only thing that’s actually disabled is the normal activation path. You can manually activate a unit that is disabled by running systemctl start; systemd won’t complain.

Many units have no installation procedure, so they can’t truly be said to be enabled or disabled; they’re just available. Such units’ status is listed as static. They only become active if activated by hand (systemctl start) or named as a dependency of other active units.

Unit files that are linked were created with systemctl link. This command creates a symbolic link from one of systemd’s system directories to a unit file that lives elsewhere in the filesystem. Such unit files can be addressed by commands or named as dependencies, but they are not full citizens of the ecosystem and have some notable quirks. For example, running systemctl disable on a linked unit file deletes the link and all references to it.

Unfortunately, the exact behavior of linked unit files is not well documented. Although the idea of keeping local unit files in a separate repository and linking them into systemd has a certain appeal, it’s probably not the best approach at this point. Just make copies.

The masked status means “administratively blocked.” systemd knows about the unit, but has been forbidden from activating it or acting on any of its configuration directives by systemctl mask. As a rule of thumb, turn off units whose status is enabled or linked with systemctl disable and reserve systemctl mask for static units.

Returning to our investigation of the cups service, we could use the following commands to reenable and start it:

```
$ sudo systemctl enable cups
Synchronizing state of cups.service with SysV init with /lib/systemd/
systemd-sysv-install...
Executing /lib/systemd/systemd-sysv-install enable cups
insserv: warning: current start runlevel(s) (empty) of script ‘cups’
    overrides LSB defaults (2 3 4 5).
insserv: warning: current stop runlevel(s) (1 2 3 4 5) of script ‘cups’
    overrides LSB defaults (1).
Created symlink from /etc/systemd/systemsockets.target.wants/cups.socket
to /lib/systemd/system/cups.socket.
Created symlink from /etc/systemd/system/multi-user.target.wants/cups.
    path to /lib/systemd/system/cups.path.
$ sudo systemctl start cups
```
Targets

Unit files can declare their relationships to other units in a variety of ways. In the example on this page, for example, the WantedBy clause says that if the system has a multi-user.target unit, that unit should depend on this one (rsync.service) when this unit is enabled.

Because units directly support dependency management, no additional machinery is needed to implement the equivalent of init’s run levels. For clarity, systemd does define a distinct class of units (of type .target) to act as well-known markers for common operating modes. However, targets have no real superpowers beyond the dependency management that’s available to any other unit.

Traditional init defines at least seven numeric run levels, but many of those aren’t actually in common use. In a perhaps-ill-advised gesture toward historical continuity, systemd defines targets that are intended as direct analogs of the init run levels (runlevel0.target, etc.). It also defines mnemonic targets for day-to-day use such as poweroff.target and graphical.target. Table 2.6 shows the mapping between init run levels and systemd targets.

**Table 2.6: Mapping between init run levels and systemd targets**

<table>
<thead>
<tr>
<th>Run level</th>
<th>Target</th>
<th>Description</th>
</tr>
</thead>
<tbody>
<tr>
<td>0</td>
<td>poweroff.target</td>
<td>System halt</td>
</tr>
<tr>
<td>emergency</td>
<td>emergency.target</td>
<td>Bare-bones shell for system recovery</td>
</tr>
<tr>
<td>1, s, single</td>
<td>rescue.target</td>
<td>Single-user mode</td>
</tr>
<tr>
<td>2</td>
<td>multi-user.target</td>
<td>Multiuser mode (command line)</td>
</tr>
<tr>
<td>3</td>
<td>multi-user.target</td>
<td>Multiuser mode with networking</td>
</tr>
<tr>
<td>4</td>
<td>multi-user.target</td>
<td>Not normally used by init</td>
</tr>
<tr>
<td>5</td>
<td>graphical.target</td>
<td>Multiuser mode with networking and GUI</td>
</tr>
<tr>
<td>6</td>
<td>reboot.target</td>
<td>System reboot</td>
</tr>
</tbody>
</table>

a. By default, multi-user.target maps to runlevel3.target, multiuser mode with networking.

The only targets to really be aware of are multi-user.target and graphical.target for day-to-day use, and rescue.target for accessing single-user mode. To change the system’s current operating mode, use the systemctl isolate command:

```
$ sudo systemctl isolate multi-user.target
```

The isolate subcommand is so-named because it activates the stated target and its dependencies but deactivates all other units.

Under traditional init, you use the telinit command to change run levels once the system is booted. Some distributions now define telinit as a symlink to the systemctl command, which recognizes how it’s being invoked and behaves appropriately.
To see the target the system boots into by default, run the `get-default` subcommand:

$ systemctl get-default
graphical.target

Most Linux distributions boot to `graphical.target` by default, which isn’t appropriate for servers that don’t need a GUI. But that’s easily changed:

$ sudo systemctl set-default multi-user.target

To see all the system’s available targets, run `systemctl list-units`:

$ systemctl list-units --type=target
Dependencies among units

Linux software packages generally come with their own unit files, so administrators don’t need a detailed knowledge of the entire configuration language. However, they do need a working knowledge of systemd’s dependency system to diagnose and fix dependency problems.

To begin with, not all dependencies are explicit. systemd takes over the functions of the old inetd and also extends this idea into the domain of the D-Bus interprocess communication system. In other words, systemd knows which network ports or IPC connection points a given service will be hosting, and it can listen for requests on those channels without actually starting the service. If a client does materialize, systemd simply starts the actual service and passes off the connection. The service runs if it’s actually used and remains dormant otherwise.

Second, systemd makes some assumptions about the normal behavior of most kinds of units. The exact assumptions vary by unit type. For example, systemd assumes that the average service is an add-on that shouldn’t be running during the early phases of system initialization. Individual units can turn off these assumptions with the line

```
DefaultDependencies=false
```

in the [Unit] section of their unit file; the default is true. See the man page for systemd.unit-type to see the exact assumptions that apply to each type of unit (e.g., man systemd.service).

A third class of dependencies are those explicitly declared in the [Unit] sections of unit files. Table 2.7 shows the available options.

<table>
<thead>
<tr>
<th>Option</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>Wants</td>
<td>Units that should be coactivated if possible, but are not required</td>
</tr>
<tr>
<td>Requires</td>
<td>Strict dependencies; failure of any prerequisite terminates this service</td>
</tr>
<tr>
<td>Requisite</td>
<td>Like Requires, but must already be active</td>
</tr>
<tr>
<td>BindsTo</td>
<td>Similar to Requires, but even more tightly coupled</td>
</tr>
<tr>
<td>PartOf</td>
<td>Similar to Requires, but affects only starting and stopping</td>
</tr>
<tr>
<td>Conflicts</td>
<td>Negative dependencies; cannot be coactive with these units</td>
</tr>
</tbody>
</table>

With the exception of Conflicts, all the options in Table 2.7 express the basic idea that the unit being configured depends on some set of other units. The exact distinctions among these options are subtle and primarily of interest to service developers. The least restrictive variant, Wants, is preferred when possible.

You can extend a unit’s Wants or Requires cohorts by creating a unit-file.wants or unit-file.requires directory in /etc/systemd/system and adding symlinks there to other unit files. Better yet, just let
systemctl do it for you. For example, the command

```
$ sudo systemctl add-wants multi-user.target my.local.service
```

adds a dependency on my.local.service to the standard multiuser target, ensuring that the service will be started whenever the system enters multiuser mode.

In most cases, such ad hoc dependencies are automatically taken care of for you, courtesy of the [Install] sections of unit files. This section includes WantedBy and RequiredBy options that are read only when a unit is enabled with systemctl enable or disabled with systemctl disable. On enablement, they make systemctl perform the equivalent of an add-wants for every WantedBy or an add-requires for every RequiredBy.

The [Install] clauses themselves have no effect in normal operation, so if a unit doesn’t seem to be started when it should be, make sure that it has been properly enabled and symlinked.
Execution order

You might reasonably guess that if unit A Requires unit B, then unit B will be started or configured before unit A. But in fact that is not the case. In systemd, the order in which units are activated (or deactivated) is an entirely separate question from that of which units to activate.

When the system transitions to a new state, systemd first traces the various sources of dependency information outlined in the previous section to identify the units that will be affected. It then uses Before and After clauses from the unit files to sort the work list appropriately. To the extent that units have no Before or After constraints, they are free to be adjusted in parallel.

Although potentially surprising, this is actually a praiseworthy design feature. One of the major design goals of systemd was to facilitate parallelism, so it makes sense that units do not acquire serialization dependencies unless they explicitly ask for them.

In practice, After clauses are typically used more frequently than Wants or Requires. Target definitions (and in particular, the reverse dependencies encoded in WantedBy and RequiredBy clauses) establish the general outlines of the services running in each operating mode, and individual packages worry only about their immediate and direct dependencies.
A more complex unit file example

Now for a closer look at a few of the directives used in unit files. Here’s a unit file for the NGINX web server, `nginx.service`:

```
[Unit]
Description=The nginx HTTP and reverse proxy server
After=network.target remote-fs.target nss-lookup.target

[Service]
Type=forking
PIDFile=/run/nginx.pid
ExecStartPre=/usr/bin/rm -f /run/nginx.pid
ExecStartPre=/usr/sbin/nginx -t
ExecStart=/usr/sbin/nginx
ExecReload=/bin/kill -s HUP $MAINPID
KillMode=process
KillSignal=SIGQUIT
TimeoutStopSec=5
PrivateTmp=true

[Install]
WantedBy=multi-user.target
```

This service is of type forking, which means that the startup command is expected to terminate even though the actual daemon continues running in the background. Since `systemd` won’t have directly started the daemon, the daemon records its PID (process ID) in the stated `PIDFile` so that `systemd` can determine which process is the daemon’s primary instance.

The `Exec` lines are commands to be run in various circumstances. `ExecStartPre` commands are run before the actual service is started; the ones shown here validate the syntax of NGINX’s configuration file and ensure that any preexisting PID file is removed. `ExecStart` is the command that actually starts the service. `ExecReload` tells `systemd` how to make the service reread its configuration file. (`systemd` automatically sets the environment variable `MAINPID` to the appropriate value.)

See [this page](#) for more information about signals.

Termination for this service is handled through `KillMode` and `KillSignal`, which tell `systemd` that the service daemon interprets a QUIT signal as an instruction to clean up and exit. The line

```
ExecStop=/bin/kill -s HUP $MAINPID
```

would have essentially the same effect. If the daemon doesn’t terminate within `TimeoutStopSec` seconds, `systemd` will force the issue by pelting it with a TERM signal and then an uncatchable KILL signal.
The PrivateTmp setting is an attempt at increasing security. It puts the service’s /tmp directory somewhere other than the actual /tmp, which is shared by all the system’s processes and users.
Local services and customizations

As you can see from the previous examples, it’s relatively trivial to create a unit file for a homegrown service. Browse the examples in /usr/lib/systemd/system and adapt one that’s close to what you want. See the man page for systemd.service for a complete list of configuration options for services. For options common to all types of units, see the page for systemd.unit.

Put your new unit file in /etc/systemd/system. You can then run

```
$ sudo systemctl enable custom.service
```

to activate the dependencies listed in the service file’s [Install] section.

As a general rule, you should never edit a unit file you didn’t write. Instead, create a configuration directory in /etc/systemd/system/unit-file.d and add one or more configuration files there called xxx.conf. The xxx part doesn’t matter; just make sure the file has a .conf suffix and is in the right location. override.conf is the standard name.

.conf files have the same format as unit files, and in fact systemd smoothes them all together with the original unit file. However, override files have priority over the original unit file should both sources try to set the value of a particular option.

One point to keep in mind is that many systemd options are allowed to appear more than once in a unit file. In these cases, the multiple values form a list and are all active simultaneously. If you assign a value in your override.conf file, that value joins the list but does not replace the existing entries. This may or may not be what you want. To remove the existing entries from a list, just assign the option an empty value before adding your own.

Let’s look at an example. Suppose that your site keeps its NGINX configuration file in a nonstandard place, say, /usr/local/www/nginx.conf. You need to run the nginx daemon with a -c /usr/local/www/nginx.conf option so that it can find the proper configuration file.

You can’t just add this option to /usr/lib/systemd/system/nginx.service because that file will be replaced whenever the NGINX package is updated or refreshed. Instead, you can use the following command sequence:

```
$ sudo mkdir /etc/systemd/system/nginx.service.d
$ sudo cat > !$/override.conf
[Service]
ExecStart=
ExecStart=/usr/sbin/nginx -c /usr/local/www/nginx.conf
<Control-D>
$ sudo systemctl daemon-reload
$ sudo systemctl restart nginx.service
```
The `>` and `!$` are shell metacharacters. The `>` redirects output to a file, and the `!$` expands to the last component of the previous command line so that you don’t have to retype it. All shells understand this notation. See [Shell basics](#) for some other handy features.

The first `ExecStart=` removes the current entry, and the second sets an alternative start command. `systemctl daemon-reload` makes `systemd` re-parse unit files. However, it does not restart daemons automatically, so you’ll also need an explicit `systemctl restart` to make the change take effect immediately.

This command sequence is such a common idiom that `systemctl` now implements it directly:

```
$ sudo systemctl edit nginx.service
<edit the override file in the editor>
$ sudo systemctl restart nginx.service
```

As shown, you must still do the `restart` by hand.

One last thing to know about override files is that they can’t modify the `[Install]` section of a unit file. Any changes you make are silently ignored. Just add dependencies directly with `systemctl add-wants` or `systemctl add-requires`.
Service and startup control caveats

systemd has many architectural implications, and adopting it is not a simple task for the teams that build Linux distributions. Current releases are mostly Frankenstein systems that adopt much of systemd but also retain a few links to the past. Sometimes the holdovers just haven’t yet been fully converted. In other cases, various forms of glue have been left behind to facilitate compatibility.

Though systemctl can and should be used for managing services and daemons, don’t be surprised when you run into traditional init scripts or their associated helper commands. If you attempt to use systemctl to disable the network on a CentOS or Red Hat system, for example, you’ll receive the following output:

```
$ sudo systemctl disable network
network.service is not a native service, redirecting to /sbin/chkconfig.
Executing /sbin/chkconfig network off
```

See this page for more information about Apache.

Traditional init scripts often continue to function on a systemd system. For example, an init script /etc/rc.d/init.d/my-old-service might be automatically mapped to a unit file such as my-old-service.service during system initialization or when systemctl daemon-reload is run. Apache 2 on Ubuntu 17.04 is a case in point: attempting to disable the apache2.service results in the following output:

```
$ sudo systemctl disable apache2
Synchronizing state of apache2.service with SysV service script with
   /lib/systemd/systemd-sysv-install.
Executing: /lib/systemd/systemd-sysv-install disable apache2
```

The end result is what you wanted, but it goes through a rather circuitous route.

Red Hat, and by extension CentOS, still run the /etc/rc.d/rc.local script at boot time if you configure it to be executable. In theory, you can use this script to hack in site-specific tweaks or post-boot tasks if desired. (At this point, though, you should really skip the hacks and do things systemd’s way by creating an appropriate set of unit files.)

Some Red Hat and CentOS boot chores continue to use config files found in the /etc/sysconfig directory. Table 2.8 summarizes these.

Table 2.8: Files and subdirectories of Red Hat’s /etc/sysconfig directory
<table>
<thead>
<tr>
<th>File or directory</th>
<th>Contents</th>
</tr>
</thead>
<tbody>
<tr>
<td>console/</td>
<td>A directory that historically allowed for custom keymapping</td>
</tr>
<tr>
<td>crond</td>
<td>Arguments to pass to the crond daemon</td>
</tr>
<tr>
<td>init</td>
<td>Configuration for handling messages from startup scripts</td>
</tr>
<tr>
<td>iptables-config</td>
<td>Loads additional iptables modules such as NAT helpers</td>
</tr>
<tr>
<td>network-scripts/</td>
<td>Accessory scripts and network config files</td>
</tr>
<tr>
<td>nfs</td>
<td>Optional RPC and NFS arguments</td>
</tr>
<tr>
<td>ntpd</td>
<td>Command-line options for ntpd</td>
</tr>
<tr>
<td>selinux</td>
<td>Symlink to /etc/selinux/config(^a)</td>
</tr>
</tbody>
</table>

a. Sets arguments for SELinux or allows you to disable it altogether

A couple of the items in **Table 2.8** merit additional comment:

- The **network-scripts** directory contains additional material related to network configuration. The only things you might need to change here are the files named *ifcfg-interface*. For example, *network-scripts/ifcfg-eth0* contains the configuration parameters for the interface eth0. It sets the interface’s IP address and networking options. See [*Red Hat and CentOS network configuration*](#) for more information about configuring network interfaces.

- The **iptables-config** file doesn’t actually allow you to modify the **iptables** (firewall) rules themselves. It just provides a way to load additional modules such as those for network address translation (NAT) if you’re going to be forwarding packets or using the system as a router. See [*this page*](#) for more information about configuring **iptables**.
systemd logging

Capturing the log messages produced by the kernel has always been something of a challenge. It became even more important with the advent of virtual and cloud-based systems, since it isn’t possible to simply stand in front of these systems’ consoles and watch what happens. Frequently, crucial diagnostic information was lost to the ether.

systemd alleviates this problem with a universal logging framework that includes all kernel and service messages from early boot to final shutdown. This facility, called the journal, is managed by the journald daemon.

System messages captured by journald are stored in the /run directory. rsyslog can process these messages and store them in traditional log files or forward them to a remote syslog server. You can also access the logs directly with the journalctl command.

Without arguments, journalctl displays all log entries (oldest first):

$ journalctl
-- Logs begin at Fri 2016-02-26 15:01:25 UTC, end at Fri 2016-02-26 15:05:16 UTC. --
Feb 26 15:01:25 ubuntu systemd-journal[285]: Runtime journal is using 4.6M (max allowed 37.0M, t
Feb 26 15:01:25 ubuntu systemd-journal[285]: Runtime journal is using 4.6M (max allowed 37.0M, t
Feb 26 15:01:25 ubuntu kernel: Initializing cgroup subsys cpuset
Feb 26 15:01:25 ubuntu kernel: Initializing cgroup subsys cpu
Feb 26 15:01:25 ubuntu kernel: Linux version 3.19.0-43-generic (build@1cy01-02) (gcc version 4.
Feb 26 15:01:25 ubuntu kernel: KERNEL supported cpus:
Feb 26 15:01:25 ubuntu kernel: Intel GenuineIntel
...

You can configure journald to retain messages from prior boots. To do this, edit /etc/systemd/journald.conf and configure the Storage attribute:

[Journal]
Storage=persistent

Once you’ve configured journald, you can obtain a list of prior boots with

$ journalctl --list-boots
-1 a73415fade0e4e7f4bea60913883d180dc Fri 2016-02-26 15:01:25 UTC
Fri 2016-02-26 15:05:16 UTC
0 0c563fa3830047ecaa2d2b053d4e661d Fri 2016-02-26 15:11:03 UTC Fri
2016-02-26 15:12:28 UTC
You can then access messages from a prior boot by referring to its index or by naming its long-form ID:

$ journalctl -b -1
$ journalctl -b a73415fade0e4e7f4bea60913883d180dc

To restrict the logs to those associated with a specific unit, use the `-u` flag:

$ journalctl -u ntp
-- Logs begin at Fri 2016-02-26 15:11:03 UTC, end at Fri 2016-02-26 15:26:07 UTC. --
Feb 26 15:11:07 ub-test-1 systemd[1]: Stopped LSB: Start NTP daemon.
Feb 26 15:11:08 ub-test-1 systemd[1]: Starting LSB: Start NTP daemon...
Feb 26 15:11:08 ub-test-1 ntpd[761]: * Starting NTP server ntpd
...

System logging is covered in more detail in Chapter 10, Logging.
2.8 FreeBSD init and Startup Scripts

FreeBSD uses a BSD-style init, which does not support the concept of run levels. To bring the system to its fully booted state, FreeBSD’s init just runs /etc/rc. This program is a shell script, but it should not be directly modified. Instead, the rc system implements a couple of standardized ways for administrators and software packages to extend the startup system and make configuration changes.

See Chapter 7 for more information about shell scripting.

/etc/rc is primarily a wrapper that runs other startup scripts, most of which live in /usr/local/etc/rc.d and /etc/rc.d. Before it runs any of those scripts, however, rc executes three files that hold configuration information for the system:

- /etc/defaults/config
- /etc/rc.conf
- /etc/rc.conf.local

These files are themselves scripts, but they typically contain only definitions for the values of shell variables. The startup scripts then check these variables to determine how to behave. (/etc/rc uses some shell magic to ensure that the variables defined in these files are visible everywhere.)

/etc/defaults/rc.conf lists all the configuration parameters and their default settings. Never edit this file, lest the startup script bogeyman hunt you down and overwrite your changes the next time the system is updated. Instead, just override the default values by setting them again in /etc/rc.conf or /etc/rc.conf.local. The rc.conf man page has an extensive list of the variables you can specify.

In theory, the rc.conf files can also specify other directories in which to look for startup scripts by your setting the value of the local_startup variable. The default value is /usr/local/etc/rc.d, and we recommend leaving it that way. For local customizations, you have the option of either creating standard rc.d-style scripts that go in /usr/local/etc/rc.d or editing the system-wide /etc/rc.local script. The former is preferred.

As you can see from peeking at /etc/rc.d, there are many different startup scripts, more than 150 on a standard installation. /etc/rc runs these scripts in the order calculated by the rcmd command, which reads the scripts and looks for dependency information that’s been encoded in a standard way.

FreeBSD’s startup scripts for garden-variety services are fairly straightforward. For example, the top of the sshd startup script is as follows:
The `rcvar` variable contains the name of a variable that’s expected to be defined in one of the `rc.conf` scripts, in this case, `sshd_enable`. If you want `sshd` (the real daemon, not the startup script; both are named `sshd`) to run automatically at boot time, put the line

`sshd_enable="YES"`

into `/etc/rc.conf`. If this variable is set to "NO" or commented out, the `sshd` script will not start the daemon or check to see whether it should be stopped when the system is shut down.

The `service` command provides a real-time interface into FreeBSD’s `rc.d` system. To stop the `sshd` service manually, for example, you could run the command

`$ sudo service sshd stop`

Note that this technique works only if the service is enabled in the `/etc/rc.conf` files. If it is not, use the subcommand `onestop`, `onestart`, or `onerestart`, depending on what you want to do. (`service` is generally forgiving and will remind you if need be, however.)
2.9 **Reboot and shutdown procedures**

Historically, UNIX and Linux machines were touchy about how they were shut down. Modern systems have become less sensitive, especially when a robust filesystem is used, but it’s always a good idea to shut down a machine nicely when possible.

Consumer operating systems of yesteryear trained many sysadmins to reboot the system as the first step in debugging any problem. It was an adaptive habit back then, but these days it more commonly wastes time and interrupts service. Focus on identifying the root cause of problems, and you’ll probably find yourself rebooting less often.

That said, it’s a good idea to reboot after modifying a startup script or making significant configuration changes. This check ensures that the system can boot successfully. If you’ve introduced a problem but don’t discover it until several weeks later, you’re unlikely to remember the details of your most recent changes.
Shutting down physical systems

The `halt` command performs the essential duties required for shutting down the system. `halt` logs the shutdown, kills nonessential processes, flushes cached filesystem blocks to disk, and then halts the kernel. On most systems, `halt -p` powers down the system as a final flourish.

`reboot` is essentially identical to `halt`, but it causes the machine to reboot instead of halting.

The `shutdown` command is a layer over `halt` and `reboot` that provides for scheduled shutdowns and ominous warnings to logged-in users. It dates back to the days of time-sharing systems and is now largely obsolete. `shutdown` does nothing of technical value beyond `halt` or `reboot`, so feel free to ignore it if you don’t have multiuser systems.
Shutting down cloud systems

You can halt or restart a cloud system either from within the server (with `halt` or `reboot`, as described in the previous section) or from the cloud provider’s web console (or its equivalent API).

Generally speaking, powering down from the cloud console is akin to turning off the power. It’s better if the virtual server manages its own shutdown, but feel free to kill a virtual server from the console if it becomes unresponsive. What else can you do?

Either way, make sure you understand what a shutdown means from the perspective of the cloud provider. It would be a shame to destroy your system when all you meant to do was reboot it.

In the AWS universe, the Stop and Reboot operations do what you’d expect. “Terminate” decommissions the instance and removes it from your inventory. If the underlying storage device is set to “delete on termination,” not only will your instance be destroyed, but the data on the root disk will also be lost. That’s perfectly fine, as long as it’s what you expect. You can enable “termination protection” if you consider this a bad thing.
2.10 Stratagems for a Nonbooting System

A variety of problems can prevent a system from booting, ranging from faulty devices to kernel upgrades gone wrong. There are three basic approaches to this situation, listed here in rough order of desirability:

- Don’t debug; just restore the system to a known-good state.
- Bring the system up just enough to run a shell, and debug interactively.
- Boot a separate system image, mount the sick system’s filesystems, and investigate from there.

The first option is the one most commonly used in the cloud, but it can be helpful on physical servers, too, as long as you have access to a recent image of the entire boot disk. If your site does backups by filesystem, a whole-system restore may be more trouble than it’s worth. We discuss the whole-system restore option in *Recovery of cloud systems*.

The remaining two approaches focus on giving you a way to access the system, identify the underlying issue, and make whatever fix is needed. Booting the ailing system to a shell is by far the preferable option, but problems that occur very early in the boot sequence may stymie this approach.

The “boot to a shell” mode is known generically as single-user mode or rescue mode. Systems that use *systemd* have an even more primitive option available in the form of emergency mode; it’s conceptually similar to single-user mode, but does an absolute minimum of preparation before starting a shell.

Because single-user, rescue, and emergency modes don’t configure the network or start network-related services, you’ll generally need physical access to the console to make use of them. As a result, single-user mode normally isn’t available for cloud-hosted systems. We review some options for reviving broken cloud images starting on *this page.*
Single-user mode

In single-user mode, also known as rescue.target on systems that use systemd, only a minimal set of processes, daemons, and services are started. The root filesystem is mounted (as is /usr, in most cases), but the network remains uninitialized.

At boot time, you request single-user mode by passing an argument to the kernel, usually single or -s. You can do this through the boot loader’s command-line interface. In some cases, it may be set up for you automatically as a boot menu option.

If the system is already running, you can bring it down to single-user mode with a shutdown (FreeBSD), telinit (traditional init), or systemctl (systemd) command.

See Chapter 3, for more information about the root account.

Sane systems prompt for the root password before starting the single-user root shell. Unfortunately, this means that it’s virtually impossible to reset a forgotten root password through single-user mode. If you need to reset the password, you’ll have to access the disk by way of separate boot media.

See Chapter 5 for more information about filesystems and mounting.

From the single-user shell, you can execute commands in much the same way as when logged in on a fully booted system. However, sometimes only the root partition is mounted; you must mount other filesystems manually to use programs that don’t live in /bin, /sbin, or /etc.

You can often find pointers to the available filesystems by looking in /etc/fstab. Under Linux, you can run fdisk -l (lowercase L option) to see a list of the local system’s disk partitions. The analogous procedure on FreeBSD is to run camcontrol devlist to identify disk devices and then run fdisk -s device for each disk.

In many single-user environments, the filesystem root directory starts off being mounted read-only. If /etc is part of the root filesystem (the usual case), it will be impossible to edit many important configuration files. To fix this problem, you’ll have to begin your single-user session by remounting / in read/write mode. Under Linux, the command

```
# mount -o rw,remount /
```

usually does the trick. On FreeBSD systems, the remount option is implicit when you repeat an existing mount, but you’ll need to explicitly specify the source device. For example,

```
# mount -o rw /dev/gpt/rootfs /
```
Single-user mode in Red Hat and CentOS is a bit more aggressive than normal. By the time you reach the shell prompt, these systems have tried to mount all local filesystems. Although this default is usually helpful, it can be problematic if you have a sick filesystem. In that case, you can boot to emergency mode by adding `systemd.unit=emergency.target` to the kernel arguments from within the boot loader (usually GRUB). In this mode, no local filesystems are mounted and only a few essential services are started.

The `fsck` command is run during a normal boot to check and repair filesystems. Depending on what filesystem you’re using for the root, you may need to run `fsck` manually when you bring the system up in single-user or emergency mode. See this page for more details about `fsck`.

Single-user mode is just a waypoint on the normal booting path, so you can terminate the single-user shell with `exit` or `<Control-D>` to continue with booting. You can also type `<Control-D>` at the password prompt to bypass single-user mode entirely.
Single-user mode on FreeBSD

FreeBSD includes a single-user option in its boot menu:

1. Boot Multi User [Enter]
2. Boot Single User
3. Escape to loader prompt
4. Reboot

Options
5. Kernel: default/kernel (1 of 2)
6. Configure Boot Options...

One nice feature of FreeBSD’s single-user mode is that it asks you what program to use as the shell. Just press <Enter> for /bin/sh.

If you choose option 3, “Escape to loader prompt,” you’ll drop into a boot-level command-line environment implemented by FreeBSD’s final-common-stage boot loader, loader.
Single-user mode with GRUB

On systems that use `systemd`, you can boot into rescue mode by appending `systemd.unit=rescue.target` to the end of the existing Linux kernel line. At the GRUB splash screen, highlight your desired kernel and press the “e” key to edit its boot options. Similarly, for emergency mode, use `systemd.unit=emergency.target`.

Here’s an example of a typical configuration:

```
linux16/vmlinuz-3.10.0-229.el7.x86_64 root=/dev/mapper/rhel_rhel-root
   ro crashkernel=auto rd.lvm.lv=rhel_rhel/swap rd.lvm.lv=rhel_rhel/
   root rfq default quiet LANG=en_US.UTF-8 systemd.unit=rescue.target
```

Type <Control-X> to start the system after you’ve made your changes.
Recovery of cloud systems

See Chapter 9 for a broader introduction to cloud computing.

It’s inherent in the nature of cloud systems that you can’t hook up a monitor or USB stick when boot problems occur. Cloud providers do what they can to facilitate problem solving, but basic limitations remain.

Backups are important for all systems, but cloud servers are particularly easy to snapshot. Providers charge extra for backups, but they’re inexpensive. Be liberal with your snapshots and you’ll always have a reasonable system image to fall back on at short notice.

From a philosophical perspective, you’re probably doing something wrong if your cloud servers require boot-time debugging. Pets and physical servers receive veterinary care when they’re sick, but cattle get euthanized. Your cloud servers are cattle; replace them with known-good copies when they misbehave. Embracing this approach helps you not only avoid critical failures but also facilitates scaling and system migration.

That said, you will inevitably need to attempt to recover cloud systems or drives, so we briefly discuss that process below.

Within AWS, single-user and emergency modes are unavailable. However, EC2 filesystems can be attached to other virtual servers if they’re backed by Elastic Block Storage (EBS) devices. This is the default for most EC2 instances, so it’s likely that you can use this method if you need to. Conceptually, it’s similar to booting from a USB drive so that you can poke around on a physical system’s boot disk.

Here’s what to do:

1. Launch a new instance in the same availability zone as the instance you’re having issues with. Ideally, this recovery instance should be launched from the same base image and should use the same instance type as the sick system.

2. Stop the problem instance. (But be careful not to “terminate” it; that operation deletes the boot disk image.)

3. With the AWS web console or CLI, detach the volume from the problem system and attach the volume to the recovery instance.

4. Log in to the recovery system. Create a mount point and mount the volume, then do whatever’s necessary to fix the issue. Then unmount the volume. (Won’t unmount? Make sure you’re not cd’ed there.)
DigitalOcean droplets offer a VNC-enabled console that you can access through the web, although the web app’s behavior is a bit wonky on some browsers. DigitalOcean does not afford a way to detach storage devices and migrate them to a recovery system the way Amazon does. Instead, most system images let you boot from an alternate recovery kernel.

To access the recovery kernel, first power off the droplet and then mount the recovery kernel and reboot. If all went well, the virtual terminal will give you access to a single-user-like mode. More detailed instructions for this process are available at digitalocean.com.

The recovery kernel is not available on all modern distributions. If you’re running a recent release and the recovery tab tells you that “The kernel for this Droplet is managed internally and cannot be changed from the control panel” you’ll need to open a support ticket with DigitalOcean to have them associate your instance with a recovery ISO, allowing you to continue your recovery efforts.

Boot issues within a Google Compute Engine instance should first be investigated by examination of the instance’s serial port information:

```bash
$ gcloud compute instances get-serial-port-output instance
```

The same information is available through GCP web console.

A disk-shuffling process similar to that described above for the Amazon cloud is also available on Google Compute Engine. You use the CLI to remove the disk from the defunct instance and boot a new instance that mounts the disk as an add-on filesystem. You can then run filesystem checks, modify boot parameters, and select a new kernel if necessary. This process is nicely detailed in Google’s documentation at [cloud.google.com/compute/docs/troubleshooting](https://cloud.google.com/compute/docs/troubleshooting).
3 Access Control and Rootly Powers

This chapter is about “access control,” as opposed to “security,” by which we mean that it focuses on the mechanical details of how the kernel and its delegates make security-related decisions. Chapter 27, Security, addresses the more general question of how to set up a system or network to minimize the chance of unwelcome access by intruders.

Access control is an area of active research, and it has long been one of the major challenges of operating system design. Over the last decade, UNIX and Linux have seen a Cambrian explosion of new options in this domain. A primary driver of this surge has been the advent of kernel APIs that allow third party modules to augment or replace the traditional UNIX access control system. This modular approach creates a variety of new frontiers; access control is now just as open to change and experimentation as any other aspect of UNIX.

Nevertheless, the traditional system remains the UNIX and Linux standard, and it’s adequate for the majority of installations. Even for administrators who want to venture into the new frontier, a thorough grounding in the basics is essential.
3.1 Standard UNIX access control

The standard UNIX access control model has remained largely unchanged for decades. With a few enhancements, it continues to be the default for general-purpose OS distributions. The scheme follows a few basic rules:

- Access control decisions depend on which user is attempting to perform an operation, or in some cases, on that user’s membership in a UNIX group.
- Objects (e.g., files and processes) have owners. Owners have broad (but not necessarily unrestricted) control over their objects.
- You own the objects you create.
- The special user account called “root” can act as the owner of any object.
- Only root can perform certain sensitive administrative operations.

Certain system calls (e.g., `settimeofday`) are restricted to root; the implementation simply checks the identity of the current user and rejects the operation if the user is not root. Other system calls (e.g., `kill`) implement different calculations that involve both ownership matching and special provisions for root. Finally, filesystems have their own access control systems, which they implement in cooperation with the kernel’s VFS layer. These are generally more elaborate than the access controls found elsewhere in the kernel. For example, filesystems are much more likely to make use of UNIX groups for access control.

See this page for more information about device files.

Complicating this picture is that the kernel and the filesystem are intimately intertwined. For example, you control and communicate with most devices through files that represent them in `/dev`. Since device files are filesystem objects, they are subject to filesystem access control semantics. The kernel uses that fact as its primary form of access control for devices.

Keep in mind that we are here describing the original design of the access control system. These days, not all of the statements above remain literally true. For example, a Linux process that bears appropriate capabilities (see this page) can now perform some operations that were previously restricted to root.
Filesystem access control

In the standard model, every file has both an owner and a group, sometimes referred to as the “group owner.” The owner can set the permissions of the file. In particular, the owner can set them so restrictively that no one else can access it. We talk more about file permissions in Chapter 5, The Filesystem (see this page).

See this page for more information about groups.

Although the owner of a file is always a single person, many people can be group owners of the file, as long as they are all part of a single group. Groups are traditionally defined in the /etc/group file, but these days group information is often stored in a network database system such as LDAP; see Chapter 17, Single Sign-On, for details.

The owner of a file gets to specify what the group owners can do with it. This scheme allows files to be shared among members of the same project.

You can determine the ownerships of a file with ls -l:

$ ls -l ~garth/todo
-rw-r----- 1 garth staff 1259 May 29 19:55 /Users/garth/todo

This file is owned by user garth and group staff. The letters and dashes in the first column symbolize the permissions on the file; see this page for details on how to decode the information. In this case, the codes mean that garth can read or write the file and that members of the staff group can read it.

See Chapter 8 for more information about the passwd and group files.

Both the kernel and the filesystem track owners and groups as numbers rather than as text names. In the most basic case, user identification numbers (UIDs for short) are mapped to usernames in the /etc/passwd file, and group identification numbers (GIDs) are mapped to group names in /etc/group. (See Chapter 17, Single Sign-On, for information about the more sophisticated options.)

The text names that correspond to UIDs and GIDs are defined only for the convenience of the system’s human users. When commands such as ls should display ownership information in a human-readable format, they must look up each name in the appropriate file or database.
Process ownership

The owner of a process can send the process signals (see this page) and can also reduce (degrade) the process’s scheduling priority. Processes actually have multiple identities associated with them: a real, effective, and saved UID; a real, effective, and saved GID; and under Linux, a “filesystem UID” that is used only to determine file access permissions. Broadly speaking, the real numbers are used for accounting (now largely vestigial), and the effective numbers are used for the determination of access permissions. The real and effective numbers are normally the same.

The saved UID and GID are parking spots for IDs that are not currently in use but that remain available for the process to invoke. The saved IDs allow a program to repeatedly enter and leave a privileged mode of operation; this precaution reduces the risk of unintended misbehavior.

See Chapter 21 for more about NFS.

The filesystem UID is generally explained as an implementation detail of NFS, the Network File System. It is usually the same as the effective UID.
The root account

The root account is UNIX’s omnipotent administrative user. It’s also known as the superuser account, although the actual username is “root”.

The defining characteristic of the root account is its UID of 0. Nothing prevents you from changing the username on this account or from creating additional accounts whose UIDs are 0; however, these are both bad ideas. (Jennine Townsend, one of our stalwart technical reviewers, commented, “Such bad ideas that I fear even mentioning them might encourage someone!”) Changes like these have a tendency to create inadvertent breaches of system security. They also create confusion when other people have to deal with the strange way you’ve configured your system.

Traditional UNIX allows the superuser (that is, any process for which the effective UID is 0) to perform any valid operation on any file or process. “Valid” is the operative word here; certain operations (such as executing a file on which the execute permission bit is not set) are forbidden even to the superuser.

Some examples of restricted operations are

- Creating device files
- Setting the system clock
- Raising resource usage limits and process priorities
- Setting the system’s hostname
- Configuring network interfaces
- Opening privileged network ports (those numbered below 1,024)
- Shutting down the system

An example of superuser powers is the ability of a process owned by root to change its UID and GID. The login program and its GUI equivalents are a case in point; the process that prompts you for your password when you log in to the system initially runs as root. If the password and username that you enter are legitimate, the login program changes its UID and GID to your UID and GID and starts up your shell or GUI environment. Once a root process has changed its ownerships to become a normal user process, it can’t recover its former privileged state.
Setuid and setgid execution

Traditional UNIX access control is complemented by an identity substitution system that’s implemented by the kernel and the filesystem in collaboration. This scheme allows specially marked executable files to run with elevated permissions, usually those of root. It lets developers and administrators set up structured ways for unprivileged users to perform privileged operations.

When the kernel runs an executable file that has its “setuid” or “setgid” permission bits set, it changes the effective UID or GID of the resulting process to the UID or GID of the file containing the program image rather than the UID and GID of the user that ran the command. The user’s privileges are thus promoted for the execution of that specific command only.

For example, users must be able to change their passwords. But since passwords are (traditionally) stored in the protected `/etc/master.passwd` or `/etc/shadow` file, users need a setuid `passwd` command to mediate their access. The `passwd` command checks to see who’s running it and customizes its behavior accordingly: users can change only their own passwords, but root can change any password.

Programs that run setuid, especially ones that run setuid to root, are prone to security problems. The setuid commands distributed with the system are theoretically secure; however, security holes have been discovered in the past and will undoubtedly be discovered in the future.

The surest way to minimize the number of setuid problems is to minimize the number of setuid programs. Think twice before installing software that needs to run setuid, and avoid using the setuid facility in your own home-grown software. Never use setuid execution on programs that were not explicitly written with setuid execution in mind.

See this page for more information about filesystem mount options.

You can disable setuid and setgid execution on individual filesystems by specifying the nosuid option to mount. It’s a good idea to use this option on filesystems that contain users’ home directories or that are mounted from less trustworthy administrative domains.
3.2 Management of the Root Account

Root access is required for system administration, and it’s also a pivot point for system security. Proper husbandry of the root account is a crucial skill.
Root account login

Since root is just another user, most systems let you log in directly to the root account. However, this turns out to be a bad idea, which is why Ubuntu forbids it by default.

To begin with, root logins leave no record of what operations were performed as root. That’s bad enough when you realize that you broke something last night at 3:00 a.m. and can’t remember what you changed; it’s even worse when an access was unauthorized and you are trying to figure out what an intruder has done to your system. Another disadvantage is that the log-in-as-root scenario leaves no record of who was actually doing the work. If several people have access to the root account, you won’t be able to tell who used it and when.

For these reasons, most systems allow root logins to be disabled on terminals, through window systems, and across the network—everywhere but on the system console. We suggest that you use these features. See PAM: cooking spray or authentication wonder? to see how to implement this policy on your particular system.

If root does have a password (that is, the root account is not disabled; see this page), that password must be of high quality. See this page for some additional comments regarding password selection.
**su: substitute user identity**

A marginally better way to access the root account is to use the `su` command. If invoked without arguments, `su` prompts for the root password and then starts up a root shell. Root privileges remain in effect until you terminate the shell by typing `<Control-D>` or the `exit` command. `su` doesn’t record the commands executed as root, but it does create a log entry that states who became root and when.

The `su` command can also substitute identities other than root. Sometimes, the only way to reproduce or debug a user’s problem is to `su` to their account so that you reproduce the environment in which the problem occurs.

If you know someone’s password, you can access that person’s account directly by executing `su - username`. As with an `su` to root, you are prompted for the password for `username`. The `-` (dash) option makes `su` spawn the shell in login mode.

The exact implications of login mode vary by shell, but login mode normally changes the number or identity of the files that the shell reads when it starts up. For example, `bash` reads `~/.bash_profile` in login mode and `~/.bashrc` in nonlogin mode. When diagnosing other users’ problems, it helps to reproduce their login environments as closely as possible by running in login mode.

On some systems, the root password allows an `su` or `login` to any account. On others, you must first `su` explicitly to root before `su`ing to another account; root can `su` to any account without entering a password.

Get in the habit of typing the full pathname to `su` (e.g., `/bin/su` or `/usr/bin/su`) rather than relying on the shell to find the command for you. This precaution gives you some protection against arbitrary programs called `su` that might have been sneaked into your search path with the intention of harvesting passwords. (For the same reason, do not include “.”, the current directory, in your shell’s search path. Although convenient, including “.” makes it easy to inadvertently run “special” versions of system commands that an intruder has left lying around as a trap. Naturally, this advice goes double for root.)

On most systems, you must be a member of the group “wheel” to use `su`.

We consider `su` to have been largely superseded by `sudo`, described in the next section. `su` is best reserved for emergencies. It’s also helpful for fixing situations in which `sudo` has been broken or misconfigured.
**sudo: limited su**

Without one of the advanced access control systems outlined starting on this page, it’s hard to enable someone to do one task (backups, for example) without giving that person free run of the system. And if the root account is used by several administrators, you really have only a vague idea of who’s using it or what they’ve done.

The most widely used solution to these problems is a program called **sudo** that is currently maintained by Todd Miller. It runs on all our example systems and is also available in source code form from sudo.ws. We recommend it as the primary method of access to the root account.

**sudo** takes as its argument a command line to be executed as root (or as another restricted user). **sudo** consults the file `/etc/sudoers` (/usr/local/etc/sudoers on FreeBSD), which lists the people who are authorized to use **sudo** and the commands they are allowed to run on each host. If the proposed command is permitted, **sudo** prompts for the user’s own password and executes the command.

Additional **sudo** commands can be executed without the “doer” having to type a password until a five-minute period (configurable) has elapsed with no further **sudo** activity. This timeout serves as a modest protection against users with **sudo** privileges who leave terminals unattended.

See Chapter 10 for more information about syslog.

**sudo** keeps a log of the command lines that were executed, the hosts on which they were run, the people who ran them, the directories from which they were run, and the times at which they were invoked. This information can be logged by syslog or placed in the file of your choice. We recommend using syslog to forward the log entries to a secure central host.

A log entry for randy’s executing **sudo /bin/cat /etc/sudoers** might look like this:

Dec 7 10:57:19 tigger sudo: randy: TTY=ttyp0 ; PWD=/tigger/users/randy;
USER=root ; COMMAND=/bin/cat /etc/sudoers

**Example configuration**

The **sudoers** file is designed so that a single version can be used on many different hosts at once. Here’s a typical example:
The first two sets of lines define groups of hosts and commands that are referred to in the permission specifications later in the file. The lists could be included literally in the specifications, but aliases make the `sudoers` file easier to read and understand; they also make the file easier to update in the future. It’s also possible to define aliases for sets of users and for sets of users as whom commands may be run.

Each permission specification line includes information about

- The users to whom the line applies
- The hosts on which the line should be heeded
- The commands that the specified users can run
- The users as whom the commands can be executed

The first permission line applies to the users mark and ed on the machines in the `PHYSICS` group (eprince, pprince, and icarus). The built-in command alias `ALL` allows them to run any command. Since no list of users is specified in parentheses, `sudo` will run commands as root.

The second permission line allows herb to run `tcpdump` on `CS` machines and dump-related commands on `PHYSICS` machines. However, the dump commands can be run only as operator, not as root. The actual command line that herb would type would be something like

```
ubuntu$ sudo -u operator /usr/sbin/tcpdump &> /dev/null
```

The user lynda can run commands as any user on any machine, except that she can’t run several common shells. Does this mean that lynda really can’t get a root shell? Of course not:

```
ubuntu$ cp -p /bin/sh /tmp/sh
ubuntu$ sudo /tmp/sh
```
Generally speaking, any attempt to allow “all commands except …” is doomed to failure, at least in a technical sense. However, it might still be worthwhile to set up the `sudoers` file this way as a reminder that root shells are strongly discouraged.

The final line allows users in group wheel to run the local `watchdog` command as root on all machines except eprince, pprince, and icarus. Furthermore, no password is required to run the command.

Note that commands in the `sudoers` file are specified with full pathnames to prevent people from executing their own programs and scripts as root. Though no examples are shown above, it is possible to specify the arguments that are permissible for each command as well.

To manually modify the `sudoers` file, use the `visudo` command, which checks to be sure no one else is editing the file, invokes an editor on it (`vi`, or whichever editor you specify in your `EDITOR` environment variable), and then verifies the syntax of the edited file before installing it. This last step is particularly important because an invalid `sudoers` file might prevent you from `sudo`ing again to fix it.

**sudo pros and cons**

The use of `sudo` has the following advantages:

- Accountability is much improved because of command logging.
- Users can do specific chores without having unlimited root privileges.
- The real root password can be known to only one or two people.
- Using `sudo` is faster than using `su` or logging in as root.
- Privileges can be revoked without the need to change the root password.
- A canonical list of all users with root privileges is maintained.
- The chance of a root shell being left unattended is lessened.
- A single file can control access for an entire network.

`sudo` has a couple of disadvantages as well. The worst of these is that any breach in the security of a sudoer’s personal account can be equivalent to breaching the root account itself. You can’t do much to counter this threat other than caution your sudoers to protect their own accounts as they would the root account. You can also run a password cracker regularly on sudoers’ passwords to ensure that they are making good password selections. All the comments on password selection from this page apply here as well.

`sudo`’s command logging can easily be subverted by tricks such as shell escapes from within an allowed program, or by `sudo sh` and `sudo su`. (Such commands do show up in the logs, so you’ll at least know they’ve been run.)
sudo vs. advanced access control

If you think of sudo as a way of subdividing the privileges of the root account, it is superior in some ways to many of the drop-in access control systems outlined starting on this page:

- You decide exactly how privileges will be subdivided. Your division can be coarser or finer than the privileges defined for you by an off-the-shelf system.
- Simple configurations—the most common—are simple to set up, maintain, and understand.
- sudo runs on all UNIX and Linux systems. You do need not worry about managing different solutions on different platforms.
- You can share a single configuration file throughout your site.
- You get consistent, high-quality logging for free.

Because the system is vulnerable to catastrophic compromise if the root account is penetrated, a major drawback of sudo-based access control is that the potential attack surface expands to include the accounts of all administrators.

sudo works well as a tool for well-intentioned administrators who need general access to root privileges. It’s also great for allowing non-administrators to perform a few specific operations. Despite a configuration syntax that suggests otherwise, it is unfortunately not a safe way to define limited domains of autonomy or to place certain operations out of bounds.

Don’t even attempt these configurations. If you need this functionality, you are much better off enabling one of the drop-in access control systems described starting on this page.

Typical setup

sudo’s configuration system has accumulated a lot of features over the years. It has also expanded to accommodate a variety of unusual situations and edge cases. As a result, the current documentation conveys an impression of complexity that isn’t necessarily warranted.

Since it’s important that sudo be reliable and secure, it’s natural to wonder if you might be exposing your systems to additional risk if you don’t make use of sudo’s advanced features and set exactly the right values for all options. The answer is no. 90% of sudoers files look something like this:

```
User_Alias  ADMINS = alice, bob, charles
ADMINS       ALL = (ALL) ALL
```

This is a perfectly respectable configuration, and in many cases there’s no need to complicate it further. We’ve mentioned a few extras you can play with in the sections below, but they’re all
problem-solving tools that are helpful for specific situations. Nothing more is required for general robustness.

Environment management

Many commands consult the values of environment variables and modify their behavior depending on what they find. In the case of commands run as root, this mechanism can be both a useful convenience and a potential route of attack.

For example, several commands run the program specified in your EDITOR environment variable to spawn a text editor. If this variable points to a hacker’s malicious program instead of an editor, it’s likely that you’ll eventually end up running that program as root. (Just to be clear, the scenario in this case is that your account has been compromised, but the attacker does not know your actual password and so cannot run **sudo** directly. Unfortunately, this is a common situation—all it takes is a terminal window left momentarily unattended.)

To minimize this risk, **sudo**’s default behavior is to pass only a minimal, sanitized environment to the commands that it runs. If your site needs additional environment variables to be passed, you can whitelist them by adding them to the **sudoers** file’s `env_keep` list. For example, the lines

```
Defaults env_keep += "SSH_AUTH_SOCK"
Defaults env_keep += "DISPLAY XAUTHORIZATION XAUTHORITY"
```

preserve several environment variables used by X Windows and by SSH key forwarding.

It’s possible to set different `env_keep` lists for different users or groups, but the configuration rapidly becomes complicated. We suggest sticking to a single, universal list and being relatively conservative with the exceptions you enshrine in the **sudoers** file.

If you need to preserve an environment variable that isn’t listed in the **sudoers** file, you can set it explicitly on the **sudo** command line. For example, the command

```
$ sudo EDITOR=emacs vipw
```

edits the system password file with **emacs**. This feature has some potential restrictions, but they’re waived for users who can run **ALL** commands.

**sudo without passwords**

It’s distressingly common to see **sudo** set up to allow command execution as root without the need to enter a password. Just for reference, that configuration is achieved with the `NOPASSWD` keyword in the **sudoers** file. For example:

```
ansible ALL = (ALL) NOPASSWD: ALL  # Don't do this
```
See Chapter 23 for more information about Ansible.

Sometimes this is done out of laziness, but more typically, the underlying need is to allow some type of unattended sudo execution. The most common cases are when performing remote configuration through a system such as Ansible, or when running commands out of cron.

Needless to say, this configuration is dangerous, so avoid it if you can. At the very least, restrict passwordless execution to a specific set of commands if you can.

See this page for more information about PAM configuration.

Another option that works well in the context of remote execution is to replace manually entered passwords with authentication through ssh-agent and forwarded SSH keys. You can configure this method of authentication through PAM on the server where sudo will actually run.

Most systems don’t include the PAM module that implements SSH-based authentication by default, but it is readily available. Look for a pam_ssh_agent_auth package.

SSH key forwarding has its own set of security concerns, but it’s certainly an improvement over no authentication at all.

**Precedence**

A given invocation of sudo might potentially be addressed by several entries in the sudoers file. For example, consider the following configuration:

```
User_Alias      ADMIN = alice, bob, charles
User_Alias      MYSQL_ADMIN = alice, bob
%wheel           ALL = (ALL) ALL
MYSQL_ADMIN      ALL = (mysql) NOPASSWD: ALL
ADMIN            ALL = (ALL) NOPASSWD: /usr/sbin/logrotate
```

Here, administrators can run the logrotate command as any user without supplying a password. MySQL administrators can run any command as mysql without a password. Anyone in the wheel group can run any command under any UID, but must authenticate with a password first.

If user alice is in the wheel group, she is potentially covered by each of the last three lines. How do you know which one will determine sudo’s behavior?

The rule is that sudo always obeys the last matching line, with matching being determined by the entire 4-tuple of user, host, target user, and command. Each of those elements must match the configuration line, or the line is simply ignored.
Therefore, `NOPASSWD` exceptions must follow their more general counterparts, as shown above. If the order of the last three lines were reversed, poor alice would have to type a password no matter what `sudo` command she attempted to run.

**sudo without a control terminal**

In addition to raising the issue of passwordless authentication, unattended execution of `sudo` (e.g., from `cron`) often occurs without a normal control terminal. There’s nothing inherently wrong with that, but it’s an odd situation that `sudo` can check for and reject if the `requiretty` option is turned on in the `sudoers` file.

This option is not the default from `sudo`’s perspective, but some OS distributions include it in their default `sudoers` files, so it’s worth checking for and removing. Look for a line of the form

```
Defaults requiretty
```

and invert its value:

```
Defaults !requiretty
```

The `requiretty` option does offer a small amount of symbolic protection against certain attack scenarios. However, it’s easy to work around and so offers little real security benefit. In our opinion, `requiretty` should be disabled as a matter of course because it is a common source of problems.

**Site-wide sudo configuration**

Because the `sudoers` file includes the current host as a matching criterion for configuration lines, you can use one master `sudoers` file throughout an administrative domain (that is, a region of your site in which hostnames and user accounts are guaranteed to be name-equivalent). This approach makes the initial `sudoers` setup a bit more complicated, but it’s a great idea, for multiple reasons. You should do it.

The main advantage of this approach is that there’s no mystery about who has what permissions on what hosts. Everything is recorded in one authoritative file. When an administrator leaves your organization, for example, there’s no need to track down all the hosts on which that user might have had `sudo` permissions. When changes are needed, you simply modify the master `sudoers` file and redistribute it.

A natural corollary of this approach is that `sudo` permissions might be better expressed in terms of user accounts rather than UNIX groups. For example,

```
%wheel        ALL = ALL
```

has some intuitive appeal, but it defers the enumeration of privileged users to each local machine. You can’t look at this line and determine who’s covered by it without an excursion to the
machine in question. Since the idea is to keep all relevant information in one place, it’s best to avoid this type of grouping option when sharing a sudoers file on a network. Of course, if your group memberships are tightly coordinated site-wide, it’s fine to use groups.

Distribution of the sudoers file is best achieved through a broader system of configuration management, as described in Chapter 23. But if you haven’t yet reached that level of organization, you can easily roll your own. Be careful, though: installing a bogus sudoers file is a quick route to disaster. This is also a good file to keep an eye on with a file integrity monitoring solution of some kind; see this page.

In the absence of a configuration management system, it’s best to use a “pull” script that runs out of cron on each host. Use scp to copy the current sudoers file from a known central repository, then validate it with visudo -c -f newsudoers before installation to verify that the format is acceptable to the local sudo. scp checks the remote server’s host key for you, ensuring that the sudoers file is coming from the host you intended and not from a spoofed server.

Hostname specifications can be a bit subtle when sharing the sudoers file. By default, sudo uses the output of the hostname command as the text to be matched. Depending on the conventions in use at your site, this name may or may not include a domain portion (e.g., anchor vs. anchor.cs.colorado.edu). In either case, the hostnames specified in the sudoers file must match the hostnames as they are returned on each host. (You can turn on thefqdn option in the sudoers file to attempt to normalize local hostnames to their fully qualified forms.)

Hostname matching gets even stickier in the cloud, where instance names often default to algorithmically generated patterns. sudo understands simple pattern-matching characters (globbing) in hostnames, so consider adopting a naming scheme that incorporates some indication of each host’s security classification from sudo’s perspective.

Alternatively, you can use your cloud provider’s virtual networking features to segregate hosts by IP address, and then match on IP addresses instead of hostnames from within the sudoers file.
Disabling the root account

If your site standardizes on the use of `sudo`, you’ll have surprisingly little use for actual root passwords. Most of your administrative team will never have occasion to use them.

That fact raises the question of whether a root password is necessary at all. If you decide that it isn’t, you can disable root logins entirely by setting root’s encrypted password to * or to some other fixed, arbitrary string. On Linux, `passwd -l` “locks” an account by prepending a ! to the encrypted password, with equivalent results.

The * and the ! are just conventions; no software checks for them explicitly. Their effect derives from their not being valid password hashes. As a result, attempts to verify root’s password simply fail.

The main effect of locking the root account is that root cannot log in, even on the console. Neither can any user successfully run `su`, because that requires a root password check as well. However, the root account continues to exist, and all the software that usually runs as root continues to do so. In particular, `sudo` works normally.

The main advantage of disabling the root account is that you needn’t record and manage root’s password. You’re also eliminating the possibility of the root password being compromised, but that’s more a pleasant side effect than a compelling reason to go passwordless. Rarely used passwords are already at low risk of violation.

It’s particularly helpful to have a real root password on physical computers (as opposed to cloud or virtual instances; see Chapters 9 and 24). Real computers are apt to require rescuing when hardware or configuration problems interfere with `sudo` or the boot process. In these cases, it’s nice to have the traditional root account available as an emergency fallback.

Ubuntu ships with the root account locked, and all administrative access is funneled through `sudo` or a GUI equivalent. If you prefer, it’s fine to set a root password on Ubuntu and then unlock the account with `sudo passwd -u root`. 
System accounts other than root

Root is generally the only user that has special status in the eyes of the kernel, but several other pseudo-users are defined by most systems. You can identify these sham accounts by their low UIDs, usually less than 100. Most often, UIDs under 10 are system accounts, and UIDs between 10 and 100 are pseudo-users associated with specific pieces of software.

See this page for more information about shadow and master.passwd.

It’s customary to replace the encrypted password field of these special users in the shadow or master.passwd file with a star so that their accounts cannot be logged in to. Their shells should be set to /bin/false or /bin/nologin as well, to protect against remote login exploits that use password alternatives such as SSH key files.

As with user accounts, most systems define a variety of system-related groups that have similarly low GIDs.

Files and processes that are part of the operating system but that need not be owned by root are sometimes assigned to the users bin or daemon. The theory was that this convention would help avoid the security hazards associated with ownership by root. It’s not a compelling argument, however, and current systems often just use the root account instead.

The main advantage of defining pseudo-accounts and pseudo-groups is that they can be used more safely than the root account to provide access to defined groups of resources. For example, databases often implement elaborate access control systems of their own. From the perspective of the kernel, they run as a pseudo-user such as “mysql” that owns all database-related resources.

See this page for more information about the nobody account.

The Network File System (NFS) uses an account called “nobody” to represent root users on other systems. For remote roots to be stripped of their rootly powers, the remote UID 0 has to be mapped to something other than the local UID 0. The nobody account acts as a generic alter ego for these remote roots. In NFSv4, the nobody account can be applied to remote users that correspond to no valid local account as well.

Since the nobody account is supposed to represent a generic and relatively powerless user, it shouldn’t own any files. If nobody does own files, remote roots can take control of them.
3.3 Extensions to the Standard Access Control Model

The preceding sections outline the major concepts of the traditional access control model. Even though this model can be summarized in a couple of pages, it has stood the test of time because it’s simple, predictable, and capable of handling the requirements of the average site. All UNIX and Linux variants continue to support this model, and it remains the default approach and the one that’s most widely used today.

As actually implemented and shipped on modern operating systems, the model includes a number of important refinements. Three layers of software contribute to the current status quo:

- The standard model as described to this point
- Extensions that generalize and fine-tune this basic model
- Kernel extensions that implement alternative approaches

These categories are not architectural layers so much as historical artifacts. Early UNIX derivatives all used the standard model, but its deficiencies were widely recognized even then. Over time, the community began to develop workarounds for a few of the more pressing issues. In the interest of maintaining compatibility and encouraging widespread adoption, changes were usually structured as refinements of the traditional system. Some of these tweaks (e.g., PAM) are now considered UNIX standards.

Over the last decade, great strides have been made toward modularization of access control systems. This evolution enables even more radical changes to access control. We review some of the more common pluggable options for Linux and FreeBSD in Modern access control.

For now, we look at some of the more prosaic extensions that are bundled with most systems. First, we consider the problems those extensions attempt to address.
Drawbacks of the standard model

Despite its elegance, the standard model has some obvious shortcomings.

- To begin with, the root account represents a potential single point of failure. If it’s compromised, the integrity of the entire system is violated, and there is essentially no limit to the damage an attacker can inflict.

- The only way to subdivide the privileges of the root account is to write setuid programs. Unfortunately, as the steady flow of security-related software updates demonstrates, it’s difficult to write secure software. Every setuid program is a potential target.

- The standard model has little to say about security on a network. No computer to which an unprivileged user has physical access can be trusted to accurately represent the ownerships of the processes it’s running. Who’s to say that someone hasn’t reformatted the disk and installed their own operating system, with UIDs of their choosing?

- In the standard model, group definition is a privileged operation. For example, there’s no way for a generic user to express the intent that only alice and bob should have access to a particular file.

- Because many access control rules are embedded in the code of individual commands and daemons (the classic example being the `passwd` program), you cannot redefine the system’s behavior without modifying the source code and recompiling. In the real world, that’s impractical and error prone.

- The standard model also has little or no support for auditing or logging. You can see which UNIX groups a user belongs to, but you can’t necessarily determine what those group memberships permit a user to do. In addition, there’s no real way to track the use of elevated privileges or to see what operations they have performed.
PAM: Pluggable Authentication Modules

See this page for more information about the `shadow` and `master.passwd` files.

User accounts are traditionally secured by passwords stored (in encrypted form) in the `/etc/shadow` or `/etc/master.passwd` file or an equivalent network database. Many programs may need to validate accounts, including `login`, `sudo`, `su`, and any program that accepts logins on a GUI workstation.

These programs really shouldn’t have hard-coded expectations about how passwords are to be encrypted or verified. Ideally, they shouldn’t even assume that passwords are in use at all. What if you want to use biometric identification, a network identity system, or some kind of two-factor authentication? Pluggable Authentication Modules to the rescue!

PAM is a wrapper for a variety of method-specific authentication libraries. Administrators specify the authentication methods they want the system to use, along with the appropriate contexts for each one. Programs that require user authentication simply call the PAM system rather than implement their own forms of authentication. PAM in turn calls the authentication library specified by the system administrator.

Strictly speaking, PAM is an authentication technology, not an access control technology. That is, instead of addressing the question “Does user X have permission to perform operation Y?”, it helps answer the precursor question, “How do I know this is really user X?”

PAM is an important component of the access control chain on most systems, and PAM configuration is a common administrative task. You can find more details on PAM in the Single Sign-On chapter starting on this page.
Kerberos: network cryptographic authentication

Like PAM, Kerberos deals with authentication rather than access control per se. But whereas PAM is an authentication framework, Kerberos is a specific authentication method. At sites that use Kerberos, PAM and Kerberos generally work together, PAM being the wrapper and Kerberos the actual implementation.

Kerberos uses a trusted third party (a server) to perform authentication for an entire network. You don’t authenticate yourself to the machine you are using, but provide your credentials to the Kerberos service. Kerberos then issues cryptographic credentials that you can present to other services as evidence of your identity.

Kerberos is a mature technology that has been in widespread use for decades. It’s the standard authentication system used by Windows, and is part of Microsoft’s Active Directory system. Read more about Kerberos here.
Filesystem access control lists

Since filesystem access control is so central to UNIX and Linux, it was an early target for elaboration. The most common addition has been support for access control lists (ACLs), a generalization of the traditional user/group/other permission model that permits permissions to be set for multiple users and groups at once.

ACLs are part of the filesystem implementation, so they have to be explicitly supported by whatever filesystem you are using. However, all major UNIX and Linux filesystems now support ACLs in one form or another.

See Chapter 21, The Network File System, for more information about NFS.

ACL support generally comes in one of two forms: an early POSIX draft standard that never quite made its way to formal adoption but was widely implemented anyway, and the system standardized by NFSv4, which adapts Microsoft Windows ACLs. Both ACL standards are described in more detail in the filesystem chapter, starting on this page.
**Linux capabilities**

Capability systems divide the powers of the root account into a handful (~30) of separate permissions.

The Linux version of capabilities derives from the defunct POSIX 1003.1e draft, which totters on despite never having been formally approved as a standard. In addition to bearing this zombie stigma, Linux capabilities raise the hackles of theorists because of nonconformance to the academic conception of a capability system. No matter; they’re here, and Linux calls them capabilities, so we will too.

Capabilities can be inherited from a parent process. They can also be enabled or disabled by attributes set on an executable file, in a process reminiscent of setuid execution. Processes can renounce capabilities that they don’t plan to use.

The traditional powers of root are simply the union of all possible capabilities, so there’s a fairly direct mapping between the traditional model and the capability model. The capability model is just more granular.

As an example, the Linux capability called CAP_NET_BIND_SERVICE controls a process’s ability to bind to privileged network ports (those numbered under 1,024). Some daemons that traditionally run as root need only this one particular superpower. In the capability world, such a daemon can theoretically run as an unprivileged user and pick up the port-binding capability from its executable file. As long as the daemon does not explicitly check to be sure that it’s running as root, it needn’t even be capability aware.

Is all this actually done in the real world? Well, no. As it happens, capabilities have evolved to become more an enabling technology than a user-facing system. They’re widely employed by higher-level systems such as AppArmor (see this page) and Docker (see Chapter 25) but are rarely used on their own.

For administrators, it’s helpful to review the capabilities(7) man page just to get a sense of what’s included in each of the capability buckets.
Linux namespaces

Linux can segregate processes into hierarchical partitions ("namespaces") from which they see only a subset of the system’s files, network ports, and processes. Among other effects, this scheme acts as a form of preemptive access control. Instead of having to base access control decisions on potentially subtle criteria, the kernel simply denies the existence of objects that are not visible from inside a given box.

Inside a partition, normal access control rules apply, and in most cases jailed processes are not even aware that they have been confined. Because confinement is irreversible, processes can run as root within a partition without fear that they might endanger other parts of the system.

This clever trick is one of the foundations of software containerization and its best-known implementation, Docker. The full system is a lot more sophisticated and includes extensions such as copy-on-write filesystem access. We have quite a bit more to say about containers in Chapter 25.

As a form of access control, namespacing is a relatively coarse approach. The construction of properly configured nests for processes to live in is also somewhat tricky. Currently, this technology is applied primarily to add-on services as opposed to intrinsic components of the operating system.
3.4 Modern access control

Given the world’s wide range of computing environments and the mixed success of efforts to advance the standard model, kernel maintainers have been reluctant to act as mediators in the larger debate over access control. In the Linux world, the situation came to a head in 2001, when the U.S. National Security Agency proposed to integrate its Security-Enhanced Linux (SELinux) system into the kernel as a standard facility.

For several reasons, the kernel maintainers resisted this merge. Instead of adopting SELinux or another, alternative system, they developed the Linux Security Modules API, a kernel-level interface that allows access control systems to integrate themselves as loadable kernel modules.

LSM-based systems have no effect unless users load them and turn them on. This fact lowers the barriers for inclusion in the standard kernel, and Linux now ships with SELinux and four other systems (AppArmor, Smack, TOMOYO, and Yama) ready to go.

Developments on the BSD side have roughly paralleled those of Linux, thanks largely to Robert Watson’s work on TrustedBSD. This code has been included in FreeBSD since version 5. It also provides the application sandboxing technology used in Apple’s macOS and iOS.

When multiple access control modules are active simultaneously, an operation must be approved by all of them to be permitted. Unfortunately, the LSM system requires explicit cooperation among active modules, and none of the current modules include this feature. For now, Linux systems are effectively limited to a choice of one LSM add-on module.
Separate ecosystems

Access control is inherently a kernel-level concern. With the exception of filesystem access control lists (see this page), there is essentially no standardization among systems with regard to alternative access control mechanisms. As a result, every kernel has its own array of available implementations, and none of them are cross-platform.

Because Linux distributions share a common kernel lineage, all Linux distributions are theoretically compatible with all the various Linux security offerings. But in practical terms, they’re not: these systems all need user-level support in the form of additional commands, modifications to user-level components, and securement profiles for daemons and services. Ergo, every distribution has only one or two access control mechanisms that it actively supports (if that).
Mandatory access control

The standard UNIX model is considered a form of “discretionary access control” because it allows the owners of access-controlled entities to set the permissions on them. For example, you might allow other users to view the contents of your home directory, or you might write a setuid program that lets other people send signals to your processes.

Discretionary access control provides no particular guarantee of security for user-level data. The downside of letting users set permissions is that users can set permissions; there’s no telling what they might do with their own files. And even with the best intentions and training, users can make mistakes.

Mandatory access control (aka MAC) systems let administrators write access control policies that override or supplement the discretionary permissions of the traditional model. For example, you might establish the rule that users’ home directories are accessible only by their owners. It then doesn’t matter if a user makes a private copy of a sensitive document and is careless with the document’s permissions; nobody else can see into that user’s home directory anyway.

MAC capabilities are an enabling technology for implementing security models such as the Department of Defense’s “multilevel security” system. In this model, security policies control access according to the perceived sensitivity of the resources being controlled. Users are assigned a security classification from a structured hierarchy. They can read and write items at the same classification level or lower but cannot access items at a higher classification. For example, a user with “secret” access can read and write “secret” objects but cannot read objects that are classified as “top secret.”

Unless you’re handling sensitive data for a government entity, it is unlikely that you will ever encounter or need to deploy such comprehensive “foreign” security models. More commonly, MAC is used to protect individual services, and it otherwise stays out of users’ way.

A well-implemented MAC policy relies on the principle of least privilege (allowing access only when necessary), much as a properly designed firewall allows only specifically recognized services and clients to pass. MAC can prevent software with code execution vulnerabilities (e.g., buffer overflows) from compromising the system by limiting the scope of the breach to the few specific resources required by that software.

MAC has unfortunately become something of a buzzword synonymous with “advanced access control.” Even FreeBSD’s generic security API is called the MAC interface, despite the fact that some plug-ins offer no actual MAC features.

Available MAC systems range from wholesale replacements for the standard model to lightweight extensions that address specific domains and use cases. The common thread among MAC implementations is that they generally add centralized, administrator-written (or vendor-supplied) policies into the access control system along with the usual mix of file permissions, access controls lists, and process attributes.
Regardless of scope, MAC represents a potentially significant departure from the standard system, and it’s one that programs expecting to deal with the standard UNIX security model may find surprising. Before committing to a full-scale MAC deployment, make sure you understand the module’s logging conventions and know how to identify and troubleshoot MAC-related problems.
Role-based access control

Another feature commonly name-checked by access control systems is role-based access control (aka RBAC), a theoretical model formalized in 1992 by David Ferraiolo and Rick Kuhn. The basic idea is to add a layer of indirection to access control calculations. Permissions, instead of being assigned directly to users, are assigned to intermediate constructs known as “roles,” and roles in turn are assigned to users. To make an access control decision, the system enumerates the roles of the current user and checks to see if any of those roles have the appropriate permissions.

Roles are similar in concept to UNIX groups, but they’re more general because they can be used outside the context of the filesystem. Roles can also have a hierarchical relationship to one another, a fact that greatly simplifies administration. For example, you might define a “senior administrator” role that has all the permissions of an “administrator” plus the additional permissions X, Y, and Z.

Many UNIX variants, including Solaris, HP-UX, and AIX, include some form of built-in RBAC system. Linux and FreeBSD have no distinct, native RBAC facility. However, it is built into several of the more comprehensive MAC options.
SELinux: Security-Enhanced Linux

SELinux is one of the oldest Linux MAC implementations and is a product of the U.S. National Security Agency. Depending on one’s perspective, that might be a source of either comfort or suspicion. (If your tastes incline toward suspicion, it’s worth noting that as part of the Linux kernel distribution, the SELinux code base is open to inspection.)

SELinux takes a maximalist approach, and it implements pretty much every flavor of MAC and RBAC one might envision. Although it has gained footholds in a few distributions, it is notoriously difficult to administer and troubleshoot. This unattributed quote from a former version of the SELinux Wikipedia page vents the frustration felt by many sysadmins:

“Intriguingly, although the stated raison d’être of SELinux is to facilitate the creation of individualized access control policies specifically attuned to organizational data custodianship practices and rules, the supportive software tools are so sparse and unfriendly that the vendors survive chiefly on “consulting,” which typically takes the form of incremental modifications to boilerplate security policies.”

Despite its administrative complexity, SELinux adoption has been slowly growing, particularly in environments such as government, finance, and health care that enforce strong and specific security requirements. It’s also a standard part of the Android platform.

Our general opinion regarding SELinux is that it’s capable of delivering more harm than benefit. Unfortunately, that harm can manifest not only as wasted time and aggravation for system administrators, but also, ironically, as security lapses. Complex models are hard to reason about, and SELinux isn’t really a level playing field; hackers that focus on it understand the system far more thoroughly than the average sysadmin.

In particular, SELinux policy development is a complicated endeavor. To protect a new daemon, for example, a policy must carefully enumerate all the files, directories, and other objects to which the process needs access. For complicated software like sendmail or httpd, this task can be quite complex. At least one company offers a three-day class on policy development.

Fortunately, many general policies are available on-line, and most SELinux-enabled distributions come with reasonable defaults. These can easily be installed and configured for your particular environment. A full-blown policy editor that aims to ease policy application can be found at seedit.sourceforge.net.

SELinux is well supported by both Red Hat (and hence, CentOS) and Fedora. Red Hat enables it by default.

Debian and SUSE Linux also have some available support for SELinux, but you must install additional packages, and the system is less aggressive in its default configuration.
Ubuntu inherits some SELinux support from Debian, but over the last few releases, Ubuntu’s focus has been on AppArmor (see this page). Some vestigial SELinux-related packages are still available, but they are generally not up to date.

/etc/selinux/config is the top-level control for SELinux. The interesting lines are

SELINUX=enforcing
SELINUXTYPE=targeted

The first line has three possible values: enforcing, permissive, or disabled. The enforcing setting ensures that the loaded policy is applied and prohibits violations. permissive allows violations to occur but logs them through syslog, which is valuable for debugging and policy development. disabled turns off SELinux entirely.

SELINUXTYPE refers to the name of the policy database to be applied. This is essentially the name of a subdirectory within /etc/selinux. Only one policy can be active at a time, and the available policy sets vary by system.

Red Hat’s default policy is targeted, which defines additional security for a few daemons that Red Hat has explicitly protected but leaves the rest of the system alone. There used to be a separate policy called strict that applied MAC to the entire system, but that policy has now been merged into targeted. Remove the unconfined and unconfineduser modules with semodule -d to achieve full-system MAC.

Red Hat also defines an mls policy that implements DoD-style multilevel security. You must install it separately with yum install selinux-policy-mls.

If you’re interested in developing your own SELinux policies, check out the audit2allow utility. It builds policy definitions from logs of violations. The idea is to permissively protect a subsystem so that its violations are logged but not enforced. You can then put the subsystem through its paces and build a policy that allows everything the subsystem actually did. Unfortunately, it’s hard to guarantee complete coverage of all code paths with this sort of ad hoc approach, so the autogenerated profiles are unlikely to be perfect.
**AppArmor**

AppArmor is a product of Canonical, Ltd., releasers of the Ubuntu distribution. It’s supported by Debian and Ubuntu, but has also been adopted as a standard by SUSE distributions. Ubuntu and SUSE enable it on default installs, although the complement of protected services is not extensive.

AppArmor implements a form of MAC and is intended as a supplement to the traditional UNIX access control system. Although any configuration is possible, AppArmor is not designed to be a user-facing system. Its main goal is service securement; that is, limiting the damage that individual programs can do if they should be compromised or run amok.

Protected programs continue to be subject to all the limitations imposed by the standard model, but in addition, the kernel filters their activities through a designated and task-specific AppArmor profile. By default, AppArmor denies all requests, so the profile must explicitly name everything the process is allowed to do. Programs without profiles, such as user shells, have no special restrictions and run as if AppArmor were not installed.

This service securement role is essentially the same configuration that’s implemented by SELinux in Red Hat’s targeted environment. However, AppArmor is designed more specifically for service securement, so it sidesteps some of the more puzzling nuances of SELinux.

AppArmor profiles are stored in `/etc/apparmor.d`, and they’re relatively readable even without detailed knowledge of the system. For example, here’s the profile for the `cups-browsed` daemon, part of the printing system on Ubuntu:

```plaintext
#include <tunables/global>

/usr/sbin/cups-browsed {

    #include <abstractations/base>
    #include <abstractations/nameservice>
    #include <abstractations/cups-client>
    #include <abstractations/dbus>
    #include <abstractations/p11-kit>

    /etc/cups/cups-browsed.conf r,
    /etc/cups/lpoptions r,
    /{var,}/run/cups/certs/* r,
    /var/cache/cups/* rw,
    /tmp/** rw,

    # Site-specific additions and overrides. See local/README.
    #include <local/usr.sbin.cups-browsed>
}
```

Most of this code is modular boilerplate. For example, this daemon needs to perform hostname lookups, so the profile interpolates `abstractations/nameservice`, which gives access to name
resolution libraries, /etc/nsswitch.conf, /etc/hosts, the network ports used with LDAP, and so on.

The profiling information that’s specific to this daemon consists (in this case) of a list of files the daemon can access, along with the permissions allowed on each file. The pattern matching syntax is a bit idiosyncratic: ** can match multiple pathname components, and {var/} matches whether var/ appears at that location or not.

Even this simple profile is quite complex under the hood. With all the #include instructions expanded, the profile is nearly 750 lines long. (And we chose this example for its brevity. Yikes!)

AppArmor refers to files and programs by pathname, which makes profiles readable and independent of any particular filesystem implementation. This approach is something of a compromise, however. For example, AppArmor doesn’t recognize hard links as pointing to the same underlying entity.
3.5 Recommended reading


4 Process Control

A process represents a running program. It’s the abstraction through which memory, processor time, and I/O resources can be managed and monitored.

It is an axiom of the UNIX philosophy that as much work as possible be done within the context of processes rather than being handled specially by the kernel. System and user processes follow the same rules, so you can use a single set of tools to control them both.
4.1 Components of a Process

A process consists of an address space and a set of data structures within the kernel. The address space is a set of memory pages that the kernel has marked for the process’s use. (Pages are the units in which memory is managed. They are usually 4KiB or 8KiB in size.) These pages contain the code and libraries that the process is executing, the process’s variables, its stacks, and various extra information needed by the kernel while the process is running. The process’s virtual address space is laid out randomly in physical memory and tracked by the kernel’s page tables.

The kernel’s internal data structures record various pieces of information about each process. Here are some of the more important of these:

- The process’s address space map
- The current status of the process (sleeping, stopped, runnable, etc.)
- The execution priority of the process
- Information about the resources the process has used (CPU, memory, etc.)
- Information about the files and network ports the process has opened
- The process’s signal mask (a record of which signals are blocked)
- The owner of the process

A “thread” is an execution context within a process. Every process has at least one thread, but some processes have many. Each thread has its own stack and CPU context but operates within the address space of its enclosing process.

Modern computer hardware includes multiple CPUs and multiple cores per CPU. A process’s threads can run simultaneously on different cores. Multithreaded applications such as BIND and Apache benefit quite a bit from this architecture because it lets them farm out requests to individual threads.

Many of the parameters associated with a process directly affect its execution: the amount of processor time it gets, the files it can access, and so on. In the following sections, we discuss the meaning and significance of the parameters that are most interesting from a system administrator’s point of view. These attributes are common to all versions of UNIX and Linux.
**PID: process ID number**

The kernel assigns a unique ID number to every process. Most commands and system calls that manipulate processes require you to specify a PID to identify the target of the operation. PIDs are assigned in order as processes are created.

Linux now defines the concept of process “namespaces,” which further restrict processes’ ability to see and affect each other. Container implementations use this feature to keep processes segregated. One side effect is that a process might appear to have different PIDs depending on the namespace of the observer. It’s kind of like Einsteinian relativity for process IDs. Refer to [Chapter 25, Containers](#), for more information.
**PPID: parent PID**

Neither UNIX nor Linux has a system call that initiates a new process running a particular program. Instead, it’s done in two separate steps. First, an existing process must clone itself to create a new process. The clone can then exchange the program it’s running for a different one.

When a process is cloned, the original process is referred to as the parent, and the copy is called the child. The PPID attribute of a process is the PID of the parent from which it was cloned, at least initially. (If the original parent dies, *init* or *systemd* becomes the new parent. See this page.)

The parent PID is a useful piece of information when you’re confronted with an unrecognized (and possibly misbehaving) process. Tracing the process back to its origin (whether that is a shell or some other program) may give you a better idea of its purpose and significance.
UID and EUID: real and effective user ID

See this page for more information about UIDs.

A process’s UID is the user identification number of the person who created it, or more accurately, it is a copy of the UID value of the parent process. Usually, only the creator (aka, the owner) and the superuser can manipulate a process.

See this page for more information about setuid execution.

The EUID is the “effective” user ID, an extra UID that determines what resources and files a process has permission to access at any given moment. For most processes, the UID and EUID are the same, the usual exception being programs that are setuid.

Why have both a UID and an EUID? Simply because it’s useful to maintain a distinction between identity and permission, and because a setuid program might not wish to operate with expanded permissions all the time. On most systems, the effective UID can be set and reset to enable or restrict the additional permissions it grants.

Most systems also keep track of a “saved UID,” which is a copy of the process’s EUID at the point at which the process first begins to execute. Unless the process takes steps to obliterate this saved UID, it remains available for use as the real or effective UID. A conservatively written setuid program can therefore renounce its special privileges for the majority of its execution and access them only at the points where extra privileges are needed.

Linux also defines a nonstandard FSUID process parameter that controls the determination of filesystem permissions. It is infrequently used outside the kernel and is not portable to other UNIX systems.
GID and EGID: real and effective group ID

See this page for more information about groups.

The GID is the group identification number of a process. The EGID is related to the GID in the same way that the EUID is related to the UID in that it can be “upgraded” by the execution of a setgid program. As with the saved UID, the kernel maintains a saved GID for each process.

The GID attribute of a process is largely vestigial. For purposes of access determination, a process can be a member of many groups at once. The complete group list is stored separately from the distinguished GID and EGID. Determinations of access permissions normally take into account the EGID and the supplemental group list, but not the GID itself.

The only time at which the GID is actually significant is when a process creates new files. Depending on how the filesystem permissions have been set, new files might default to adopting the GID of the creating process. See this page for details.
Niceness

A process’s scheduling priority determines how much CPU time it receives. The kernel computes priorities with a dynamic algorithm that takes into account the amount of CPU time that a process has recently consumed and the length of time it has been waiting to run. The kernel also pays attention to an administratively set value that’s usually called the “nice value” or “niceness,” so called because it specifies how nice you are planning to be to other users of the system. We discuss niceness in detail here.
Most nondaemon processes have an associated control terminal. The control terminal determines the default linkages for the standard input, standard output, and standard error channels. It also distributes signals to processes in response to keyboard events such as `<Control-C>`; see the discussion starting here.

Of course, actual terminals are rare outside of computer museums these days. Nevertheless, they live on in the form of pseudo-terminals, which are still widely used throughout UNIX and Linux systems. When you start a command from the shell, for example, your terminal window typically becomes the process’s control terminal.
4.2 The Life Cycle of a Process

To create a new process, a process copies itself with the `fork` system call. `fork` creates a copy of the original process, and that copy is largely identical to the parent. The new process has a distinct PID and has its own accounting information. (Technically, Linux systems use `clone`, a superset of `fork` that handles threads and includes additional features. `fork` remains in the kernel for backward compatibility but calls `clone` behind the scenes.)

`fork` has the unique property of returning two different values. From the child’s point of view, it returns zero. The parent receives the PID of the newly created child. Since the two processes are otherwise identical, they must both examine the return value to figure out which role they are supposed to play.

After a `fork`, the child process often uses one of the `exec` family of routines to begin the execution of a new program. These calls change the program that the process is executing and reset the memory segments to a predefined initial state. The various forms of `exec` differ only in the ways in which they specify the command-line arguments and environment to be given to the new program.

When the system boots, the kernel autonomously creates and installs several processes. The most notable of these is `init` or `systemd`, which is always process number 1. This process executes the system’s startup scripts, although the exact manner in which this is done differs slightly between UNIX and Linux. All processes other than the ones the kernel creates are descendants of this primordial process. See Chapter 2, Booting and System Management Daemons, for more information about booting and the various flavors of `init` daemon.

`init` (or `systemd`) also plays another important role in process management. When a process completes, it calls a routine named `_exit` to notify the kernel that it is ready to die. It supplies an exit code (an integer) that tells why it’s exiting. By convention, zero indicates a normal or “successful” termination.

Before a dead process can be allowed to disappear completely, the kernel requires that its death be acknowledged by the process’s parent, which the parent does with a call to `wait`. The parent receives a copy of the child’s exit code (or if the child did not exit voluntarily, an indication of why it was killed) and can also obtain a summary of the child’s resource use if it wishes.

This scheme works fine if parents outlive their children and are conscientious about calling `wait` so that dead processes can be disposed of. If a parent dies before its children, however, the kernel recognizes that no `wait` is forthcoming. The kernel adjusts the orphan processes to make them children of `init` or `systemd`, which politely performs the `wait` needed to get rid of them when they die.
Signals

Signals are process-level interrupt requests. About thirty different kinds are defined, and they’re used in a variety of ways:

- They can be sent among processes as a means of communication.
- They can be sent by the terminal driver to kill, interrupt, or suspend processes when keys such as \texttt{<Control-C>} and \texttt{<Control-Z>} are pressed.
- They can be sent by an administrator (with \texttt{kill}) to achieve various ends.
- They can be sent by the kernel when a process commits an infraction such as division by zero.
- They can be sent by the kernel to notify a process of an “interesting” condition such as the death of a child process or the availability of data on an I/O channel.

When a signal is received, one of two things can happen. If the receiving process has designated a handler routine for that particular signal, the handler is called with information about the context in which the signal was delivered. Otherwise, the kernel takes some default action on behalf of the process. The default action varies from signal to signal. Many signals terminate the process; some also generate core dumps (if core dumps have not been disabled; a core dump is a copy of a process’s memory image, which is sometimes useful for debugging).

Specifying a handler routine for a signal is referred to as catching the signal. When the handler completes, execution restarts from the point at which the signal was received.

To prevent signals from arriving, programs can request that they be either ignored or blocked. A signal that is ignored is simply discarded and has no effect on the process. A blocked signal is queued for delivery, but the kernel doesn’t require the process to act on it until the signal has been explicitly unblocked. The handler for a newly unblocked signal is called only once, even if the signal was received several times while reception was blocked.

Table 4.1 lists some signals that administrators should be familiar with. The uppercase convention for the names derives from C language tradition. You might also see signal names written with a SIG prefix (e.g., SIGHUP) for similar reasons.

Table 4.1: Signals every administrator should know
Other signals, not shown in Table 4.1, mostly report obscure errors such as “illegal instruction.” The default handling for such signals is to terminate with a core dump. Catching and blocking are generally allowed because some programs are smart enough to try to clean up whatever problem caused the error before continuing.

The BUS and SEGV signals are also error signals. We’ve included them in the table because they’re so common: when a program crashes, it’s usually one of these two signals that finally brings it down. By themselves, the signals are of no specific diagnostic value. Both of them indicate an attempt to use or access memory improperly.

The signals named KILL and STOP cannot be caught, blocked, or ignored. The KILL signal destroys the receiving process, and STOP suspends its execution until a CONT signal is received. CONT can be caught or ignored, but not blocked.

TSTP is a “soft” version of STOP that might be best described as a request to stop. It’s the signal generated by the terminal driver when <Control-Z> is typed on the keyboard. Programs that catch this signal usually clean up their state, then send themselves a STOP signal to complete the stop operation. Alternatively, programs can ignore TSTP to prevent themselves from being stopped from the keyboard.

The signals KILL, INT, TERM, HUP, and QUIT all sound as if they mean approximately the same thing, but their uses are actually quite different. It’s unfortunate that such vague terminology was selected for them. Here’s a decoding guide:

- KILL is unblockable and terminates a process at the kernel level. A process can never actually receive or handle this signal.
- **INT** is sent by the terminal driver when the user presses `<Control-C>`. It’s a request to terminate the current operation. Simple programs should quit (if they catch the signal) or simply allow themselves to be killed, which is the default if the signal is not caught. Programs that have interactive command lines (such as shells) should stop what they’re doing, clean up, and wait for user input again.

- **TERM** is a request to terminate execution completely. It’s expected that the receiving process will clean up its state and exit.

- **HUP** has two common interpretations. First, it’s understood as a reset request by many daemons. If a daemon is capable of rereading its configuration file and adjusting to changes without restarting, a HUP can generally trigger this behavior.

  Second, HUP signals are sometimes generated by the terminal driver in an attempt to “clean up” (i.e., kill) the processes attached to a particular terminal. This behavior is largely a holdover from the days of wired terminals and modem connections, hence the name “hangup.”

  Shells in the C shell family (**tcsh** et al.) usually make background processes immune to HUP signals so that they can continue to run after the user logs out. Users of Bourne-ish shells (**ksh**, **bash**, etc.) can emulate this behavior with the **nohup** command.

- **QUIT** is similar to **TERM**, except that it defaults to producing a core dump if not caught. A few programs cannibalize this signal and interpret it to mean something else.

The signals **USR1** and **USR2** have no set meaning. They’re available for programs to use in whatever way they’d like. For example, the Apache web server interprets a HUP signal as a request for an immediate restart. A USR1 signal initiates a more graceful transition in which existing client conversations are allowed to finish.
kill: send signals

As its name implies, the `kill` command is most often used to terminate a process. `kill` can send any signal, but by default it sends a TERM. `kill` can be used by normal users on their own processes or by root on any process. The syntax is

```
kill [-signal] pid
```

where `signal` is the number or symbolic name of the signal to be sent (as shown in Table 4.1) and `pid` is the process identification number of the target process.

A `kill` without a signal number does not guarantee that the process will die, because the TERM signal can be caught, blocked, or ignored. The command

```
$ kill -9 pid
```

“guarantees” that the process will die because signal 9, KILL, cannot be caught. Use `kill -9` only if a polite request fails. We put quotes around “guarantees” because processes can on occasion become so wedged that even KILL does not affect them, usually because of some degenerate I/O vapor lock such as waiting for a volume that has disappeared. Rebooting is usually the only way to get rid of these processes.

`killall` kills processes by name. For example, the following command kills all Apache web server processes:

```
$ sudo killall httpd
```

The `pkill` command searches for processes by name (or other attributes, such as EUID) and sends the specified signal. For example, the following command sends a TERM signal to all processes running as the user ben:

```
$ sudo pkill -u ben
```
Process and thread states

As you saw in the previous section, a process can be suspended with a STOP signal and returned to active duty with a CONT signal. The state of being suspended or runnable applies to the process as a whole and is inherited by all the process’s threads. (Individual threads can in fact be managed similarly. However, those facilities are primarily of interest to developers; system administrators needn’t concern themselves.)

Even when nominally runnable, threads must often wait for the kernel to complete some background work for them before they can continue execution. For example, when a thread reads data from a file, the kernel must request the appropriate disk blocks and then arrange for their contents to be delivered into the requesting process’s address space. During this time, the requesting thread enters a short-term sleep state in which it is ineligible to execute. Other threads in the same process can continue to run, however.

You’ll sometimes see entire processes described as “sleeping” (for example, in ps output—see the next section). Since sleeping is a thread-level attribute, this convention is a bit deceptive. A process is generally reported as “sleeping” when all its threads are asleep. Of course, the distinction is moot in the case of single-threaded processes, which remain the most common case.

Interactive shells and system daemons spend most of their time sleeping, waiting for terminal input or network connections. Since a sleeping thread is effectively blocked until its request has been satisfied, its process generally receives no CPU time unless it receives a signal or a response to one of its I/O requests.

See this page for more information about hard-mounting NFS filesystems.

Some operations can cause processes or threads to enter an uninterruptible sleep state. This state is usually transient and is not observed in ps output (denoted by a D in the STAT column; see Table 4.2). However, a few degenerate situations can cause it to persist. The most common cause involves server problems on an NFS filesystem mounted with the hard option. Since processes in the uninterruptible sleep state cannot be roused even to service a signal, they cannot be killed. To get rid of them, you must fix the underlying problem or reboot.

In the wild, you might occasionally see “zombie” processes that have finished execution but that have not yet had their status collected by their parent process (or by init or systemd). If you see zombies hanging around, check their PPIDs with ps to find out where they’re coming from.
4.3 **ps: Monitor Processes**

The `ps` command is the system administrator’s main tool for monitoring processes. Although versions of `ps` differ in their arguments and display, they all deliver essentially the same information. Part of the enormous variation among versions of `ps` can be traced back to differences in the development history of UNIX. However, `ps` is also a command that vendors tend to customize for other reasons. It’s closely tied to the kernel’s handling of processes, so it tends to reflect all a vendor’s underlying kernel changes.

`ps` can show the PID, UID, priority, and control terminal of processes. It also informs you how much memory a process is using, how much CPU time it has consumed, and what its current status is (running, stopped, sleeping, etc.). Zombies show up in a `ps` listing as `<exiting>` or `<defunct>`.

Implementations of `ps` have become hopelessly complex over the years. Several vendors have abandoned the attempt to define meaningful displays and made their `pses` completely configurable. With a little customization work, almost any desired output can be produced.

As a case in point, the `ps` used by Linux is a highly polymorphous version that understands option sets from multiple historical lineages. Almost uniquely among UNIX commands, Linux’s `ps` accepts command-line flags with or without dashes but might assign different interpretations to those forms. For example, `ps -a` is not the same as `ps a`.

Do not be alarmed by all this complexity: it’s there mainly for developers, not for system administrators. Although you will use `ps` frequently, you only need to know a few specific incantations.

You can obtain a useful overview of all the processes running on the system with `ps aux`. The `a` option says show all processes, and `x` says show even processes that don’t have a control terminal; `u` selects the “user oriented” output format. Here’s an example of `ps aux` output on a machine running Red Hat:
Command names in brackets are not really commands at all but rather kernel threads scheduled as processes. The meaning of each field is shown in Table 4.2.

**Table 4.2: Explanation of ps aux output**
Another useful set of arguments is `lax`, which gives more technical information. The `a` and `x` options are as above (show every process), and `l` selects the “long” output format. `ps lax` might be slightly faster to run than `ps aux` because it doesn’t have to translate every UID to a username — efficiency can be important if the system is already bogged down.

Shown here in an abbreviated example, `ps lax` includes fields such as the parent process ID (PPID), niceness (NI), and the type of resource on which the process is waiting (WCHAN, short for “wait channel”).
Commands with long argument lists may have the command-line output cut off. Add `w` to the list of flags to display more columns in the output. Add `w` twice for unlimited column width, handy for those processes that have exceptionally long command-line arguments, such as some `java` applications.

Administrators frequently need to identify the PID of a process. You can find the PID by `grep`ping the output of `ps`:

```
$ ps aux | grep sshd
root 6811 0.0 0.0 78056 1340 ? Ss 16:04 0:00 /usr/sbin/sshd
```

Note that the `ps` output includes the `grep` command itself, since the `grep` was active in the process list at the time `ps` was running. You can remove this line from the output with `grep -v`:

```
$ ps aux | grep -v grep | grep sshd
root 6811 0.0 0.0 78056 1340 ? Ss 16:04 0:00 /usr/sbin/sshd
```

You can also determine the PID of a process with the `pidof` command:

```
$ pidof /usr/sbin/sshd
6811
```

Or with the `pgrep` utility:

```
$ pgrep sshd
6811
```

`pidof` and `pgrep` show all processes that match the passed string. We often find a simple `grep` to offer the most flexibility, though it can be a bit more verbose.
4.4 Interactive Monitoring with top

Commands like `ps` show you a snapshot of the system as it was at the time. Often, that limited sample is insufficient to convey the big picture of what’s really going on. `top` is a sort of real-time version of `ps` that gives a regularly updated, interactive summary of processes and their resource usage. For example:

```
redhat$ top
```

```
top - 20:07:43 up 1:59,  3 users,  load average: 0.45, 0.16, 0.09
Tasks: 251 total,  1 running, 250 sleeping,   0 stopped,   0 zombie
%CPU(s):  0.7 us, 1.2 sy,  0.0 ni, 98.0 id,  0.0 wa,  0.0 hi,  0.2 si,  0.0 st
KiB Mem : 1013672 total, 128394 free, 547176 used, 338192 buff/cache
KiB Swap: 2097148 total, 2089188 free, 7960 used. 242556 avail Mem

   PID USER   PR NI  VIRT  RES   SHR  %CPU  %MEM    TIME+  COMMAND
 2731 root    20   0  193316 34848 15184  S   1.7  3.4  0:30.39  Xorg
 25721 ulsah  20   0  619412 27216 17636  S   1.0  2.7  0:03.67  konsole
 25296 ulsah  20   0  260724 6068  3268  S   0.7  0.6  0:17.78  prlcc
  747 root    20   0  4372   604  504  S   0.3  0.1  0:02.68  rmgd
  846 root    20   0  141744  384  192  S   0.3  0.0  0:01.74  prltoolsd
 1647 root    20   0  177436 3656  2632  S   0.3  0.4  0:04.47  cupsd
10246 ulsah  20   0  130156 1936 1256  R   0.3  0.2  0:00.10  top
   1 root    20   0  59620  5472  3348  S   0.0  0.5  0:02.09  systemd
   2 root    20   0   0   0   0  S   0.0  0.0  0:00.02  kthreadd
   3 root    20   0   0   0   0  S   0.0  0.0  0:00.03  ksoftirqd/0
   9 root    20   0   0   0   0  S   0.0  0.0  0:00.00  rcuob/0
...

   7 root    rt   0   0   0   0  S   0.0  0.0  0:00.20  migration/0
   8 root    20   0   0   0   0  S   0.0  0.0  0:00.00  rcu_bh
   5 root    0 -20   0   0   0  S   0.0  0.0  0:00.00  kworker/0:+
```

By default, the display updates every 1–2 seconds, depending on the system. The most CPU-consumptive processes appear at the top. `top` also accepts input from the keyboard to send signals and `renice` processes (see the next section). You can then observe how your actions affect the overall condition of the machine.

---

Learn how to interpret the CPU, memory, and load details in Chapter 29.

---

The summary information in the first few lines of `top` output is one of the first places to look at to analyze the health of the system. It shows a condensed view of the system load, memory usage, number of processes, and a breakdown of how the CPU is being used.

On multicore systems, CPU usage is an average of all the cores in the system. Under Linux, press 1 (numeral one) while `top` is open to switch to a display of the individual cores. On FreeBSD, run `top -P` to achieve the same effect.
On FreeBSD systems, you can set the TOP environment variable to pass additional arguments to top. We recommend -H to show all threads for multithreaded processes rather than just a summary, plus -P to display all CPU cores. Add export TOP="-HP" to your shell initialization file to make these changes persistent between shell sessions.

Root can run top with the -q option to goose it up to the highest possible priority. This option can be useful when you are trying to track down a process that has already brought the system to its knees.

We also like htop, an open source, cross-platform, interactive process viewer that offers more features and has a nicer interface than that of top. It is not yet available as a package for our example systems, but you can download a binary or source version from the developer’s web site at hisham.hm/htop.
4.5 NICE AND RENICE: INFLUENCE SCHEDULING PRIORITY

The “niceness” of a process is a numeric hint to the kernel about how the process should be treated in relation to other processes contending for the CPU. The strange name is derived from the fact that it determines how nice you are going to be to other users of the system. A high niceness means a low priority for your process: you are going to be nice. A low or negative value means high priority: you are not very nice.

It’s highly unusual to set priorities by hand these days. On the puny systems where UNIX originated, performance was significantly affected by which process was on the CPU. Today, with more than adequate CPU power on every desktop, the scheduler does a good job of managing most workloads. The addition of scheduling classes gives developers additional control when fast response is essential.

The range of allowable niceness values varies among systems. In Linux the range is -20 to +19, and in FreeBSD it’s -20 to +20.

Unless the user takes special action, a newly created process inherits the niceness of its parent process. The owner of the process can increase its niceness but cannot lower it, even to return the process to the default niceness. This restriction prevents processes running at low priority from bearing high-priority children. However, the superuser can set nice values arbitrarily.

I/O performance has not kept up with increasingly fast CPUs. Even with today’s high-performance SSDs, disk bandwidth remains the primary bottleneck on most systems. Unfortunately, a process’s niceness has no effect on the kernel’s management of its memory or I/O; high-nice processes can still monopolize a disproportionate share of these resources.

A process’s niceness can be set at the time of creation with the nice command and adjusted later with the renice command. nice takes a command line as an argument, and renice takes a PID or (sometimes) a username.

Some examples:

$ nice -n 5 ~/bin/longtask  // Lowers priority (raise nice) by 5
$ sudo renice -5 8829      // Sets niceness to -5
$ sudo renice 5 -u boggs   // Sets niceness of boggs’s proc to 5

Unfortunately, there is little agreement among systems about how the desired priorities should be specified; in fact, even nice and renice from the same system usually don’t agree. To complicate things, a version of nice is built into the C shell and some other common shells (but not bash). If you don’t type the full path to nice, you’ll get the shell’s version rather than the operating system’s. To sidestep this ambiguity, we suggest using the fully qualified path to the system’s version, found at /usr/bin/nice.
Table 4.3 summarizes the variations. A *prio* is an absolute niceness, while an *incr* is relative to the niceness of the shell from which *nice* or *renice* is run. Only the shell *nice* understands plus signs (in fact, it requires them); leave them out in all other circumstances.

**Table 4.3: How to express priorities for nice and renice**

<table>
<thead>
<tr>
<th>System</th>
<th>Range</th>
<th>OS nice</th>
<th>csh nice</th>
<th>renice</th>
</tr>
</thead>
<tbody>
<tr>
<td>Linux</td>
<td>-20 to 19</td>
<td>-n incr</td>
<td>+incr or -incr</td>
<td><em>prio</em> or -n <em>prio</em></td>
</tr>
<tr>
<td>FreeBSD</td>
<td>-20 to 20</td>
<td>-n incr</td>
<td>+incr or -incr</td>
<td>incr or -n incr</td>
</tr>
</tbody>
</table>
4.6 The /proc Filesystem

The Linux versions of `ps` and `top` read their process status information from the `/proc` directory, a pseudo-filesystem in which the kernel exposes a variety of interesting information about the system’s state.

Despite the name `/proc` (and the name of the underlying filesystem type, “proc”), the information is not limited to process information—a variety of status information and statistics generated by the kernel are represented here. You can even modify some parameters by writing to the appropriate `/proc` file. See this page for some examples.

Although a lot of the information is easiest to access through front-end commands such as `vmstat` and `ps`, some of the more obscure nuggets must be read directly from `/proc`. It’s worth poking around in this directory to familiarize yourself with everything that’s there. `man proc` has a comprehensive explanation of its contents.

Because the kernel creates the contents of `/proc` files on the fly (as they are read), most appear to be empty, 0-byte files when listed with `ls -l`. You’ll have to `cat` or `less` the contents to see what they actually contain. But be cautious—a few files contain or link to binary data that can confuse your terminal emulator if viewed directly.

Process-specific information is divided into subdirectories named by PID. For example, `/proc/1` is always the directory that contains information about `init`. Table 4.4 lists the most useful per-process files.

<table>
<thead>
<tr>
<th>File</th>
<th>Contents</th>
</tr>
</thead>
<tbody>
<tr>
<td>cgroup</td>
<td>The control groups to which the process belongs</td>
</tr>
<tr>
<td>cmd</td>
<td>Command or program the process is executing</td>
</tr>
<tr>
<td>cmdline</td>
<td>Complete command line of the process (null-separated)</td>
</tr>
<tr>
<td>cwd</td>
<td>Symbolic link to the process's current directory</td>
</tr>
<tr>
<td>environ</td>
<td>The process's environment variables (null-separated)</td>
</tr>
<tr>
<td>exe</td>
<td>Symbolic link to the file being executed</td>
</tr>
<tr>
<td>fd</td>
<td>Subdirectory containing links for each open file descriptor</td>
</tr>
<tr>
<td>fdinfo</td>
<td>Subdirectory containing further info for each open file descriptor</td>
</tr>
<tr>
<td>maps</td>
<td>Memory mapping information (shared segments, libraries, etc.)</td>
</tr>
<tr>
<td>ns</td>
<td>Subdirectory with links to each namespace used by the process.</td>
</tr>
<tr>
<td>root</td>
<td>Symbolic link to the process's root directory (set with chroot)</td>
</tr>
<tr>
<td>stat</td>
<td>General process status information (best decoded with <code>ps</code>)</td>
</tr>
<tr>
<td>statm</td>
<td>Memory usage information</td>
</tr>
</tbody>
</table>

a. Might be unavailable if the process is swapped out of memory
The individual components contained within the `cmdline` and `environ` files are separated by null characters rather than newlines. You can filter their contents through `tr "\000" "\n"` to make them more readable.

The `fd` subdirectory represents open files in the form of symbolic links. File descriptors that are connected to pipes or network sockets don’t have an associated filename. The kernel supplies a generic description as the link target instead.

The `maps` file can be useful for determining what libraries a program is linked to or depends on.

 FreeBSD includes a similar-but-different implementation of `/proc`. However, its use has been deprecated because of neglect in the code base and a history of security issues. It’s still available for compatibility but is not mounted by default. To mount it, use the command

```
freebsd$ sudo mount -t procfs proc /proc
```

(To automatically mount the `/proc` filesystem at boot time, append the line `proc /proc procfs rw 0 0` to `/etc/fstab`.)

The filesystem layout is similar—but not identical—to the Linux version of procfs. The information for a process includes its status, a symbolic link to the file being executed, details about the process’s virtual memory, and other low-level information. See also `man procfs`. 
4.7 STRACE AND TRUSS: TRACE SIGNALS AND SYSTEM CALLS

It’s often difficult to figure out what a process is actually doing. The first step is generally to make an educated guess based on indirect data collected from the filesystem, logs, and tools such as ps.

If those sources of information prove insufficient, you can snoop on the process at a lower level with the strace (Linux; usually an optional package) or truss (FreeBSD) command. These commands display every system call that a process makes and every signal it receives. You can attach strace or truss to a running process, snoop for a while, and then detach from the process without disturbing it. (Well, usually. strace can interrupt system calls. The monitored process must then be prepared to restart them. This is a standard rule of UNIX software hygiene, but it’s not always observed.)

Although system calls occur at a relatively low level of abstraction, you can usually tell quite a bit about a process’s activity from the call trace. For example, the following log was produced by strace run against an active copy of top (which was running as PID 5810):

```
redhat$ sudo strace -p 5810
gettimeofday([{1116193514}, {300, 0}) = 0
open("/proc", O_RDONLY|O_NONBLOCK|O_LARGEFILE|O_DIRECTORY) = 7
fstat64(7, {st_mode=S_IFDIR|0555, st_size=0, ...}) = 0
fcnt64(7, F_SETFD, FD_CLOEXEC) = 0
gedents64(7, /* 36 entries */ , 1024) = 1016
gedents64(7, /* 39 entries */ , 1024) = 1016
stat64("/proc/1", {st_mode=S_IFDIR|0555, st_size=0, ...}) = 0
open("/proc/1/stat", O_RDONLY) = 8
read(8, "1 (init) S 0 0 0 0 -1 4194560 73", ..., 1023) = 191
close(8) = 0
...
```

Not only does strace show you the name of every system call made by the process, but it also decodes the arguments and shows the result code that the kernel returns.

In the example above, top starts by checking the current time. It then opens and stats the /proc directory and reads the directory’s contents, thereby obtaining a list of running processes. top goes on to stat the directory representing the init process and then opens /proc/1/stat to read init’s status information.

System call output can often reveal errors that are not reported by the process itself. For example, filesystem permission errors or socket conflicts are usually quite obvious in the output of strace or truss. Look for system calls that return error indications, and check for nonzero values.

strace is packed with goodies, most of which are documented in the man page. For example, the -f flag follows forked processes. This feature is useful for tracing daemons (such as httpd) that
spawn many children. The `-e trace=file` option displays only file-related operations. This feature is especially handy for discovering the location of evasive configuration files.

Here’s a similar example from FreeBSD that uses `truss`. In this case, we trace how `cp` copies a file:

```
freebsd$ truss cp /etc/passwd /tmp/pw
...
lstat("/etc/passwd", { mode=-rw-r--r-- ,inode=13576, size=2380,
blksize=4096 }) = 0 (0x0)
umask(0x1ff) = 18 (0x12)
umask(0x12) = 511 (0x1ff)
fstatat(AT_FDCWD,"/etc/passwd", { mode=-rw-r--r-- ,inode=13576,
size=2380, blksize=4096 }, 0x0) = 0 (0x0)
stat("/tmp/pw", 0x7fffffff440) ERR#2 'No such file or directory'
openat(AT_FDCWD,"/etc/passwd", O_RDONLY, 00) = 3 (0x3)
openat(AT_FDCWD,"/tmp/pw", O_WRONLY|O_CREAT, 0100644) = 4 (0x4)
mmap(0x0,2380, PROT_READ, MAP_SHARED, 3, 0x0) = 34366304256
(0x8000630000)
write4("$FreeBSD: releng/11.0/etc/mast"..., 2380) = 2380 (0x94c)
close(4) = 0 (0x0)
close(3) = 0 (0x0)
...
```

After allocating memory and opening library dependencies (not shown), `cp` uses the `lstat` system call to check the current status of the `/etc/passwd` file. It then runs `stat` on the path of the prospective copy, `/tmp/pw`. That file does not yet exist, so the `stat` fails and `truss` decodes the error for you as “No such file or directory.”

`cp` then invokes the `openat` system call (with the O_RDONLY option) to read the contents of `/etc/passwd`, followed by an `openat` of `/tmp/pw` with O_WRONLY to create the new destination file. It then maps the contents of `/etc/passwd` into memory (with `mmap`) and writes out the data with `write`. Finally, `cp` cleans up after itself by closing both file handles.

System call tracing is a powerful debugging tool for administrators. Turn to these tools after more traditional routes such as examining log files and configuring a process for verbose output have been exhausted. Do not be intimidated by the dense output; it’s usually sufficient to focus on the human-readable portions.
4.8 Runaway processes

“Runaway” processes are those that soak up significantly more of the system’s CPU, disk, or network resources than their usual role or behavior would lead you to expect. Sometimes, such programs have their own bugs that have led to degenerate behavior. In other cases, they fail to deal appropriately with upstream failures and get stuck in maladaptive loops. For example, a process might reattempt the same failing operation over and over again, flooring the CPU. In yet another category of cases, there is no bug per se, but the software is simply inefficient in its implementation and greedy with the system’s resources.

All these situations merit investigation by a system administrator, not only because the runaway process is most likely malfunctioning but also because it typically interferes with the operation of other processes that are running on the system.

The line between pathological behavior and normal behavior under heavy workload is vague. Often, the first step in diagnosis is to figure out which of these phenomena you are actually observing. Generally, system processes should always behave reasonably, so obvious misbehavior on the part of one of these processes is automatically suspicious. User processes such as web servers and databases might simply be overloaded.

You can identify processes that are using excessive CPU time by looking at the output of `ps` or `top`. Also check the system load averages as reported by the `uptime` command. Traditionally, these values quantify the average number of processes that have been runnable over the previous 1-, 5-, and, 15-minute intervals. Under Linux, the load average also takes account of busyness caused by disk traffic and other forms of I/O.

For CPU bound systems, the load averages should be less than the total number of CPU cores available on your system. If they are not, the system is overloaded. Under Linux, check total CPU utilization with `top` or `ps` to determine whether high load averages relate to CPU load or to I/O. If CPU utilization is near 100%, that is probably the bottleneck.

Processes that use excessive memory relative to the system’s physical RAM can cause serious performance problems. You can check the memory size of processes by running `top`. The VIRT column shows the total amount of virtual memory allocated by each process, and the RES column shows the portion of that memory currently mapped to specific memory pages (the “resident set”).

Both of these numbers can include shared resources such as libraries and thus are potentially misleading. A more direct measure of process-specific memory consumption is found in the DATA column, which is not shown by default. To add this column to `top`’s display, type the `f` key once `top` is running and select DATA from the list by pressing the space bar. The DATA value indicates the amount of memory in each process’s data and stack segments, so it’s relatively specific to individual processes (modulo shared memory segments). Look for growth
over time as well as absolute size. On FreeBSD, SIZE is the equivalent column and is shown by default.

Make a concerted effort to understand what’s going on before you terminate a seemingly runaway process. The best route to debugging the issue and preventing a recurrence is to have a live example you can investigate. Once you kill a misbehaving process, most of the available evidence disappears.

Keep the possibility of hacking in mind as well. Malicious software is typically not tested for correctness in a variety of environments, so it’s more likely than average to enter some kind of degenerate state. If you suspect misfeasance, obtain a system call trace with `strace` or `truss` to get a sense of what the process is doing (e.g., cracking passwords) and where its data is stored.

Runaway processes that produce output can fill up an entire filesystem, causing numerous problems. When a filesystem fills up, lots of messages will be logged to the console and attempts to write to the filesystem will produce error messages.

The first thing to do in this situation is to determine which filesystem is full and which file is filling it up. The `df -h` command shows filesystem disk use in human-readable units. Look for a filesystem that’s 100% or more full. (Most filesystem implementations reserve ~5% of the storage space for “breathing room,” but processes running as root can encroach on this space, resulting in a reported usage that is greater than 100%). Use the `du -h` command on the identified filesystem to determine which directory is using the most space. Rinse and repeat with `du` until the large files are discovered.

`df` and `du` report disk usage in subtly different manners. `df` reports the disk space used by a mounted filesystem according to disk block totals in the filesystem’s metadata. `du` sums the sizes of all files in a given directory. If a file is unlinked (deleted) from the filesystem but is still referenced by some running process, `df` reports the space but `du` does not. This disparity persists until the open file descriptor is closed or the file is truncated. If you can’t determine which process is using a file, try running the `fuser` and `lsof` commands (covered in detail [here](#)) to get more information.
4.9 Periodic processes

It’s often useful to have a script or command executed without any human intervention. Common use cases include scheduled backups, database maintenance activities, or the execution of nightly batch jobs. As is typical of UNIX and Linux, there’s more than one way to achieve this goal.
cron: schedule commands

The cron daemon is the traditional tool for running commands on a predetermined schedule. It starts when the system boots and runs as long as the system is up. There are multiple implementations of cron, but fortunately for administrators, the syntax and functionality of the various versions is nearly identical.

For reasons that are unclear, cron has been renamed crond on Red Hat. But it is still the same cron we all know and love.

cron reads configuration files containing lists of command lines and times at which they are to be invoked. The command lines are executed by sh, so almost anything you can do by hand from the shell can also be done with cron. If you prefer, you can even configure cron to use a different shell.

A cron configuration file is called a “crontab,” short for “cron table.” Crontabs for individual users are stored under /var/spool/cron (Linux) or /var/cron/tabs (FreeBSD). There is at most one crontab file per user. Crontab files are plain text files named with the login names of the users to whom they belong. cron uses these filenames (and the file ownership) to figure out which UID to use when running the commands contained in each file. The crontab command transfers crontab files to and from this directory.

cron tries to minimize the time it spends reparsing configuration files and making time calculations. The crontab command helps maintain cron’s efficiency by notifying cron when crontab files change. Ergo, you shouldn’t edit crontab files directly, because this approach might result in cron not noticing your changes. If you do get into a situation where cron doesn’t seem to acknowledge a modified crontab, a HUP signal sent to the cron process forces it to reload on most systems.

See Chapter 10 for more information about syslog.

cron normally does its work silently, but most versions can keep a log file (usually /var/log/cron) that lists the commands that were executed and the times at which they ran. Glance at the cron log file if you’re having problems with a cron job and can’t figure out why.

The format of crontab files

All the crontab files on a system share a similar format. Comments are introduced with a pound sign (#) in the first column of a line. Each non-comment line contains six fields and represents one command:

minute hour dom month weekday command
The first five fields tell **cron** when to run the *command*. They’re separated by whitespace, but within the *command* field, whitespace is passed along to the shell. The fields in the time specification are interpreted as shown in **Table 4.5**. An entry in a crontab is colloquially known as a “cron job.”

**Table 4.5: Crontab time specifications**

<table>
<thead>
<tr>
<th>Field</th>
<th>Description</th>
<th>Range</th>
</tr>
</thead>
<tbody>
<tr>
<td>minute</td>
<td>Minute of the hour</td>
<td>0 to 59</td>
</tr>
<tr>
<td>hour</td>
<td>Hour of the day</td>
<td>0 to 23</td>
</tr>
<tr>
<td>dom</td>
<td>Day of the month</td>
<td>1 to 31</td>
</tr>
<tr>
<td>month</td>
<td>Month of the year</td>
<td>1 to 12</td>
</tr>
<tr>
<td>weekday</td>
<td>Day of the week</td>
<td>0 to 6 (0 = Sunday)</td>
</tr>
</tbody>
</table>

Each of the time-related fields can contain

- A star, which matches everything
- A single integer, which matches exactly
- Two integers separated by a dash, matching a range of values
- A range followed by a slash and a step value, e.g., 1-10/2
- A comma-separated list of integers or ranges, matching any value

For example, the time specification

```
45 10 * * 1-5
```

means “10:45 a.m., Monday through Friday.” A hint: never use stars in every field unless you want the command to be run every minute, which is useful only in testing scenarios. One minute is the finest granularity available to **cron** jobs.

Time ranges in crontabs can include a step value. For example, the series 0,3,6,9,12,15,18 can be written more concisely as 0-18/3. You can also use three-letter text mnemonics for the names of months and days, but not in combination with ranges. As far as we know, this feature works only with English names.

There is a potential ambiguity to watch out for with the *weekday* and *dom* fields. Every day is both a day of the week and a day of the month. If both *weekday* and *dom* are specified, a day need satisfy only one of the two conditions to be selected.

For example,
means “every half-hour on Friday, and every half-hour on the 13th of the month,” not “every half-hour on Friday the 13th.”

The command is the sh command line to be executed. It can be any valid shell command and should not be quoted. The command is considered to continue to the end of the line and can contain blanks or tabs.

Percent signs (%) indicate newlines within the command field. Only the text up to the first percent sign is included in the actual command. The remaining lines are given to the command as standard input. Use a backslash (\) as an escape character in commands that have a meaningful percent sign, for example, date +\%

Although sh is involved in executing the command, the shell does not act as a login shell and does not read the contents of ~/.profile or ~/.bash_profile. As a result, the command’s environment variables might be set up somewhat differently from what you expect. If a command seems to work fine when executed from the shell but fails when introduced into a crontab file, the environment is the likely culprit. If need be, you can always wrap your command with a script that sets up the appropriate environment variables.

We also suggest using the fully qualified path to the command, ensuring that the job will work properly even if the PATH is not set as expected. For example, the following command logs the date and uptime to a file in the user’s home directory every minute:

```
* * * * * echo $(/bin/date) - $(/usr/bin/uptime) >> ~/uptime.log
```

Alternatively, you can set environment variables explicitly at the top of the crontab:

```
PATH=/bin:/usr/bin
* * * * * echo $(date) - $(uptime) >> ~/uptime.log
```

Here are a few more examples of valid crontab entries:

```
*/10 * * * 1,3,5 echo ruok | /usr/bin/nc localhost 2181 |
mail -s "TCP port 2181 status" ben@admin.com
```

This line emails the results of a connectivity check on port 2181 every 10 minutes on Mondays, Wednesdays, and Fridays. Since cron executes command by way of sh, special shell characters like pipes and redirects function as expected.

```
0 4 * * * Sun (/usr/bin/mysqldump --user=maintenance --optimize --all-databases)
```

This entry runs the mysqldump maintenance program on Sundays at 4:00 a.m. Since the output is not saved to a file or otherwise discarded, it will be emailed to the owner of the crontab.
find /tmp -mtime +7 -type f -exec rm -f {} ';' 

This command runs at 1:20 each morning. It removes all files in the /tmp directory that have not been modified in 7 days. The ';' at the end of the line marks the end of the subcommand arguments to find.

`cron` does not try to compensate for commands that are missed while the system is down. However, it is smart about time adjustments such as shifts into and out of daylight saving time.

If your cron job is a script, be sure to make it executable (with `chmod +x`) or `cron` won’t be able to execute it. Alternatively, set the cron command to invoke a shell on your script directly (e.g., `bash -c ~/bin/myscript.sh`).

Crontab management

`crontab filename` installs `filename` as your crontab, replacing any previous version. `crontab -e` checks out a copy of your crontab, invokes your editor on it (as specified by the EDITOR environment variable), and then resubmits it to the crontab directory. `crontab -l` lists the contents of your crontab to standard output, and `crontab -r` removes it, leaving you with no crontab file at all.

Root can supply a `username` argument to edit or view other users’ crontabs. For example, `crontab -r jsmith` erases the crontab belonging to the user jsmith, and `crontab -e jsmith` edits it. Linux allows both a `username` and a `filename` argument in the same command, so the username must be prefixed with `-u` to disambiguate (e.g., `crontab -u jsmith crontab.new`).

Without command-line arguments, most versions of `crontab` try to read a crontab from standard input. If you enter this mode by accident, don’t try to exit with <Control-D>; doing so erases your entire crontab. Use <Control-C> instead. FreeBSD requires you to supply a dash as the `filename` argument to make `crontab` pay attention to its standard input. Smart.

Many sites have experienced subtle but recurrent network glitches that occur because administrators have configured `cron` to run the same command on hundreds of machines at exactly the same time, causing delays or excessive load. Clock synchronization with NTP exacerbates the problem. This issue is easy to fix with a random delay script.

`cron` logs its activities through syslog using the facility “cron,” with most messages submitted at level “info.” Default syslog configurations generally send `cron` log data to its own file.

Other crontabs

In addition to looking for user-specific crontabs, `cron` also obeys system crontab entries found in `/etc/crontab` and in the `/etc/cron.d` directory. These files have a slightly different format from the per-user crontab files: they allow commands to be run as an arbitrary user. An extra `username` field comes before the command name. The `username` field is not present in garden-variety crontab files because the crontab’s `filename` supplies this same information.
In general, `/etc/crontab` is a file for system administrators to maintain by hand, whereas `/etc/cron.d` is a sort of depot into which software packages can install any crontab entries they might need. Files in `/etc/cron.d` are by convention named after the packages that install them, but `cron` doesn't care about or enforce this convention.

Linux distributions also pre-install crontab entries that run the scripts in a set of well-known directories, thereby providing another way for software packages to install periodic jobs without any editing of a crontab file. For example, scripts in `/etc/cron.hourly`, `/etc/cron.daily`, and `/etc/cron.weekly` are run hourly, daily, and weekly, respectively.

**cron access control**

Two config files specify which users may submit crontab files. For Linux, the files are `/etc/cron.{allow,deny}`, and on FreeBSD they are `/var/cron/{allow,deny}`. Many security standards require that crontabs be available only to service accounts or to users with a legitimate business need. The `allow` and `deny` files facilitate compliance with these requirements.

If the `cron.allow` file exists, then it contains a list of all users that may submit crontabs, one per line. No unlisted person can invoke the `crontab` command. If the `cron.allow` file doesn’t exist, then the `cron.deny` file is checked. It, too, is just a list of users, but the meaning is reversed: everyone except the listed users is allowed access.

If neither the `cron.allow` file nor the `cron.deny` file exists, systems default (apparently at random, there being no dominant convention) either to allowing all users to submit crontabs or to limiting crontab access to root. In practice, a starter configuration is typically included in the default OS installation, so the question of how `crontab` might behave without configuration files is moot. Most default configurations allow all users to access `cron` by default.

It’s important to note that on most systems, access control is implemented by `crontab`, not by `cron`. If a user is able to sneak a crontab file into the appropriate directory by other means, `cron` will blindly execute the commands it contains. Therefore it is vital to maintain root ownership of `/var/spool/cron` and `/var/cron/tabs`. OS distributions always set the permissions correctly by default.
systemd timers

See Chapter 2, for an introduction to systemd and units.

In accordance with its mission to duplicate the functions of all other Linux subsystems, systemd includes the concept of timers, which activate a given systemd service on a predefined schedule. Timers are more powerful than crontab entries, but they are also more complicated to set up and manage. Some Linux distributions (e.g., CoreOS) have abandoned cron entirely in favor of systemd timers, but our example systems all continue to include cron and to run it by default.

We have no useful advice regarding the choice between systemd timers and crontab entries. Use whichever you prefer for any given task. Unfortunately, you do not really have the option to standardize on one system or the other, because software packages add their jobs to a random system of their own choice. You’ll always have to check both systems when you are trying to figure out how a particular job gets run.

Structure of systemd timers

A systemd timer comprises two files:

- A timer unit that describes the schedule and the unit to activate
- A service unit that specifies the details of what to run

In contrast to crontab entries, systemd timers can be described both in absolute calendar terms (“Wednesdays at 10:00 a.m.”) and in terms that are relative to other events (“30 seconds after system boot”). The options combine to allow powerful expressions that don’t suffer the same constraints as cron jobs. Table 4.6 describes the time expression options.

<table>
<thead>
<tr>
<th>Type</th>
<th>Time basis</th>
</tr>
</thead>
<tbody>
<tr>
<td>OnActiveSec</td>
<td>Relative to the time at which the timer itself is activated</td>
</tr>
<tr>
<td>OnBootSec</td>
<td>Relative to system boot time</td>
</tr>
<tr>
<td>OnStartupSec</td>
<td>Relative to the time at which systemd was started</td>
</tr>
<tr>
<td>OnUnitActiveSec</td>
<td>Relative to the time the specified unit was last active</td>
</tr>
<tr>
<td>OnUnitInactiveSec</td>
<td>Relative to the time the specified unit was last inactive</td>
</tr>
<tr>
<td>OnCalendar</td>
<td>A specific day and time</td>
</tr>
</tbody>
</table>

As their names suggest, values for these timer options are given in seconds. For example, OnActiveSec=30 is 30 seconds after the timer activates. The value can actually be any valid systemd time expression, as discussed in more detail starting here.
**systemd timer example**

Red Hat and CentOS include a preconfigured **systemd** timer that cleans up the system’s temporary files once a day. Below, we take a more detailed look at an example. First, we enumerate all the defined timers with the **systemctl** command. (We rotated the output table below to make it readable. Normally, each timer produces one long line of output.)

```
redhat$ systemctl list-timers
NEXT Sun 2017-06-18 10:24:33 UTC
LEFT 18h left
LAST Sat 2017-06-17 00:45:29 UTC
PASSED 15h ago
UNIT systemd-tmpfiles-clean.timer
ACTIVATES systemd-tmpfiles-clean.service
```

The output lists both the name of the timer unit and the name of the service unit it activates. Since this is a default system timer, the unit file lives in the standard systemd unit directory, `/usr/lib/systemd/system`. Here’s the timer unit file:

```
redhat$ cat /usr/lib/systemd/system/systemd-tmpfiles-clean.timer
[Unit]
Description=Daily Cleanup of Temporary Directories
[Timer]
OnBootSec=15min
OnUnitActiveSec=1d
```

The timer first activates 15 minutes after boot and then fires once a day thereafter. Note that some kind of trigger for the initial activation (here, `OnBootSec`) is always necessary. There is no single specification that achieves an “every X minutes” effect on its own.

Astute observers will notice that the timer does not actually specify which unit to run. By default, **systemd** looks for a service unit that has the same name as the timer. You can specify a target unit explicitly with the **Unit** option.

In this case, the associated service unit holds no surprises:

```
redhat$ cat /usr/lib/systemd/system/systemd-tmpfiles-clean.service
[Unit]
Description=Cleanup of Temporary Directories
DefaultDependencies=no
Conflicts=shutdown.target
After=systemd-readahead-collect.service systemd-readahead-replay.
    service local-fs.target time-sync.target
Before=shutdown.target

[Service]
Type=simple
ExecStart=/usr/bin/systemd-tmpfiles --clean
IOSchedulingClass=idle
```
You can run the target service directly (that is, independently of the timer) with `systemctl start systemctl-tmpfiles-clean`, just like any other service. This fact greatly facilitates the debugging of scheduled tasks, which can be a source of much administrative anguish when you are using `cron`.

To create your own timer, drop `.timer` and `.service` files in `/etc/systemd/system`. If you want the timer to run at boot, add

```
[Install]
WantedBy=multi-user.target
```

to the end of the timer’s unit file. Don’t forget to enable the timer at boot time with `systemctl enable`. (You can also start the timer immediately with `systemctl start`.)

A timer’s AccuracySec option delays its activation by a random amount of time within the specified time window. This feature is handy when a timer runs on a large group of networked machines and you want to avoid having all the timers fire at exactly the same moment. (Recall that with `cron`, you need to use a random delay script to achieve this feat.)

AccuracySec defaults to 60 seconds. If you want your timer to execute at exactly the scheduled time, use `AccuracySec=1ns`. (A nanosecond is probably close enough. Note that you won’t actually obtain nanosecond accuracy.)

**systemd time expressions**

Timers allow for flexible specification of dates, times, and intervals. The `systemd.time` man page is the authoritative reference for the specification grammar.

You can use interval-valued expressions instead of seconds for relative timings such as those used as the values of `OnActiveSec` and `OnBootSec`. For example, the following forms are all valid:

```
OnBootSec=2h 1m
OnStartupSec=1week 2days 3hours
OnActiveSec=1hr20m30sec10msec
```

Spaces are optional in time expressions. The minimum granularity is nanoseconds, but if your timer fires too frequently (more than once every two seconds) `systemd` temporarily disables it.

In addition to triggering at periodic intervals, timers can be scheduled to activate at specific times by including the `OnCalendar` option. This feature offers the closest match to the syntax of a traditional `cron` job, but its syntax is more expressive and flexible. Table 4.7 shows some examples of time specifications that could be used as the value of `OnCalendar`.

<table>
<thead>
<tr>
<th>Time specification</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>2017-07-04</td>
<td>July 4th, 2017 at 00:00:00 (midnight)</td>
</tr>
</tbody>
</table>
Fri-Mon *-7-4  July 4th each year, but only if it falls on Fri–Mon
Mon-Wed *-* 12:00:00 Mondays, Tuesdays, and Wednesdays at noon
Mon 17:00:00  Mondays at 5:00 p.m.
weekly  Mondays at 00:00:00 (midnight)
monthly  The 1st day of the month at 00:00:00 (midnight)
*:0/10  Every 10 minutes, starting at the 0th minute
*-*-* 11/12:10:0 At 11:10 and 23:10 every day

In time expressions, stars are placeholders that match any plausible value. As in crontab files, slashes introduce an increment value. The exact syntax is a bit different from that used in crontabs, however: crontabs want the incremented object to be a range (e.g., 9-17/2, “every two hours between 9:00 a.m. and 5:00 p.m.”), but systemd time expressions take only a start value (e.g., 9/2, “every two hours starting at 9:00 a.m.”).

**Transient timers**

You can use the **systemd-run** command to schedule the execution of a command according to any of the normal systemd timer types, but without creating task-specific timer and service unit files. For example, to pull a Git repository every ten minutes:

```
$ systemd-run --on-calendar '*:0/10' /bin/sh -c "cd /app && git pull"
```

Running timer as unit run-8823.timer.
Will run service as unit run-8823.service.

**systemd** returns a transient unit identifier that you can list with **systemctl**. (Once again, we futzed with the output format below...)

```
$ systemctl list-timers run-8823.timer
NEXT  Sat 2017-06-17 20:40:07 UTC
LEFT  9min left
LAST  Sat 2017-06-17 20:30:07 UTC
PASSED  18s ago
```

```
$ systemctl list-units run-8823.timer
UNIT  run-8823.timer
LOAD  loaded
ACTIVE  active
SUB  waiting
DESCRIPTION /bin/sh -c "cd /app && git pull"
```

To cancel and remove a transient timer, just stop it by running **systemctl stop**:

```
$ sudo systemctl stop run-8823.timer
```

**systemd-run** functions by creating timer and unit files for you in subdirectories of **/run/systemd/system**. However, transient timers do not persist after a reboot. To make them permanent, you can fish them out of **/run**, tweak them as necessary, and install them in
/etc/systemd/system. Be sure to stop the transient timer before starting or enabling the permanent version.
Common uses for scheduled tasks

In this section, we look at a couple of common chores that are often automated through cron or systemd.

Sending mail

The following crontab entry implements a simple email reminder. You can use an entry like this to automatically email the output of a daily report or the results of a command execution. (Lines have been folded to fit the page. In reality, this is one long line.)

```
30 4 25 * * /usr/bin/mail -s "Time to do the TPS reports"
    ben@admin.com
    TPS reports are due at the end of the month! Get
    busy!%Sincerely,%cron%
```

Note the use of the % character both to separate the command from the input text and to mark line endings within the input. This entry sends email at 4:30 a.m. on the 25th day of each month.

Cleaning up a filesystem

When a program crashes, the kernel may write out a file (usually named core.pid, core, or program.core) that contains an image of the program’s address space. Core files are useful for developers, but for administrators they are usually a waste of space. Users often don’t know about core files, so they tend not to disable their creation or delete them on their own. You can use a cron job to clean up these core files or other vestiges left behind by misbehaving and crashed processes.

Rotating a log file

Systems vary in the quality of their default log file management, and you will probably need to adjust the defaults to conform to your local policies. To “rotate” a log file means to divide it into segments by size or by date, keeping several older versions of the log available at all times. Since log rotation is a recurrent and regularly occurring event, it’s an ideal task to be scheduled. See Management and rotation of log files, for more details.

Running batch jobs

Some long-running calculations are best run as batch jobs. For example, messages can accumulate in a queue or database. You can use a cron job to process all the queued messages at once as an ETL (extract, transform, and load) to another location, such as a data warehouse.

Some databases benefit from routine maintenance. For example, the open source distributed database Cassandra has a repair function that keeps the nodes in a cluster in sync. These maintenance tasks are good candidates for execution through cron or systemd.
**Backing up and mirroring**

You can use a scheduled task to automatically back up a directory to a remote system. We suggest running a full backup once a week, with incremental differences each night. Run backups late at night when the load on the system is likely to be low.

Mirrors are byte-for-byte copies of filesystems or directories that are hosted on another system. They can be used as a form of backup or as a way to make files available at more than one location. Web sites and software repositories are often mirrored to offer better redundancy and to offer faster access for users that are physically distant from the primary site. Use periodic execution of the **rsync** command to maintain mirrors and keep them up to date.
Quick: which of the following would you expect to find in a “filesystem”?

- Processes
- Audio devices
- Kernel data structures and tuning parameters
- Interprocess communication channels

If the system is UNIX or Linux, the answer is “all the above, and more!” And yes, you might find some files in there, too. (It’s perhaps more accurate to say that these entities are *represented* within the filesystem. In most cases, the filesystem is used as a rendezvous point to connect clients with the drivers they are seeking.)

The basic purpose of a filesystem is to represent and organize the system’s storage resources. However, programmers have been eager to avoid reinventing the wheel when it comes to managing other types of objects. It has often proved convenient to map these objects into the filesystem namespace. This unification has some advantages (consistent programming interface, easy access from the shell) and some disadvantages (filesystem implementations suggestive of Frankenstein’s monster), but like it or not, this is the UNIX (and hence, the Linux) way.

The filesystem can be thought of as comprising four main components:

- A namespace – a way to name things and organize them in a hierarchy
- An API – a set of system calls for navigating and manipulating objects
- Security models – schemes for protecting, hiding, and sharing things
- An implementation – software to tie the logical model to the hardware

Modern kernels define an abstract interface that accommodates many different back-end filesystems. Some portions of the file tree are handled by traditional disk-based implementations. Others are fielded by separate drivers within the kernel. For example, network filesystems are handled by a driver that forwards the requested operations to a server on another computer.

Unfortunately, the architectural boundaries are not clearly drawn, and quite a few special cases exist. For example, “device files” define a way for programs to communicate with drivers inside the kernel. Device files are not really data files, but they’re handled through the filesystem and their characteristics are stored on disk.

Another complicating factor is that the kernel supports more than one type of disk-based filesystem. The predominant standards are the ext4, XFS, and UFS filesystems, along with Oracle’s ZFS and Btrfs. However, many others are available, including, Veritas’s VxFS and JFS from IBM.

“Foreign” filesystems are also widely supported, including the FAT and NTFS filesystems used by Microsoft Windows and the ISO 9660 filesystem used on older CD-ROMs.

The filesystem is a rich topic that we approach from several different angles. This chapter tells where to find things on your system and describes the characteristics of files, the meanings of permission bits, and the use of some basic commands that view and set attributes. Chapter 20, Storage, is where you’ll find the more technical filesystem topics such as disk partitioning.

Chapter 21, The Network File System, describes NFS, a file sharing system that is commonly used for remote file access between UNIX and Linux systems. Chapter 22, SMB, describes an analogous system from the Windows world.

See the sections starting here for more information about specific filesystems.

With so many different filesystem implementations available, it may seem strange that this chapter reads as if there were only a single filesystem. We can be vague about the underlying code because most modern filesystems either try to implement the traditional filesystem functionality in a faster and more reliable manner, or they add extra features as a layer on top of the standard filesystem semantics. Some filesystems do both. For better or worse, too much existing software depends on the model described in this chapter for that model to be discarded.
5.1 Pathnames

The filesystem is presented as a single unified hierarchy that starts at the directory / and continues downward through an arbitrary number of subdirectories. / is also called the root directory. This single-hierarchy system differs from the one used by Windows, which retains the concept of partition-specific namespaces.

Graphical user interfaces often refer to directories as “folders,” even on Linux systems. Folders and directories are exactly the same thing; “folder” is just linguistic leakage from the worlds of Windows and macOS. Nevertheless, it’s worth noting that the word “folder” tends to raise the hackles of some techies. Don’t use it in technical contexts unless you’re prepared to receive funny looks.

The list of directories that must be traversed to locate a particular file plus that file’s filename form a pathname. Pathnames can be either absolute (e.g., /tmp/foo) or relative (e.g., book4/filesystem). Relative pathnames are interpreted starting at the current directory. You might be accustomed to thinking of the current directory as a feature of the shell, but every process has one.

The terms filename, pathname, and path are more or less interchangeable—or at least, we use them interchangeably in this book. Filename and path can be used for both absolute and relative paths; pathname usually suggests an absolute path.

The filesystem can be arbitrarily deep. However, each component of a pathname (that is, each directory) must have a name no more than 255 characters long. There’s also a limit on the total path length you can pass into the kernel as a system call argument (4,095 bytes on Linux, 1,024 bytes on BSD). To access a file with a pathname longer than this, you must cd to an intermediate directory and use a relative pathname.
5.2 Filesystem Mounting and Unmounting

The filesystem is composed of smaller chunks—also called filesystems—each of which consists of one directory and its subdirectories and files. It’s normally apparent from context which type of “filesystem” is being discussed, but for clarity in the following discussion, we use the term “file tree” to refer to the overall layout and reserve the word “filesystem” for the branches attached to the tree.

Some filesystems live on disk partitions or on logical volumes backed by physical disks, but as mentioned earlier, filesystems can be anything that obeys the proper API: a network file server, a kernel component, a memory-based disk emulator, etc. Most kernels have a nifty “loop” filesystem that lets you mount individual files as if they were distinct devices. It’s useful for mounting DVD-ROM images stored on disk or for developing filesystem images without having to worry about repartitioning. Linux systems can even treat existing portions of the file tree as filesystems. This trick lets you duplicate, move, or hide portions of the file tree.

In most situations, filesystems are attached to the tree with the `mount` command. `mount` maps a directory within the existing file tree, called the mount point, to the root of the newly attached filesystem. The previous contents of the mount point become temporarily inaccessible as long as another filesystem is mounted there. Mount points are usually empty directories, however.

For example,

```
$ sudo mount /dev/sda4 /users
```

installs the filesystem stored on the disk partition represented by `/dev/sda4` under the path `/users`. You could then use `ls /users` to see that filesystem’s contents.

On some systems, `mount` is a just a wrapper that calls filesystem-specific commands such as `mount.ntfs` or `mount_smbfs`. You’re free to call these helper commands directly if you need to; they sometimes offer additional options that the `mount` wrapper does not understand. On the other hand, the generic `mount` command suffices for day-to-day use.

You can run the `mount` command without any arguments to see all the filesystems that are currently mounted. On Linux systems, there might be 30 or more, most of which represent various interfaces to the kernel.

The `/etc/fstab` file lists filesystems that are normally mounted on the system. The information in this file allows filesystems to be automatically checked (with `fsck`) and mounted (with `mount`) at boot time, with options you specify. The `fstab` file also serves as documentation for the layout of the filesystems on disk and enables short commands such as `mount /usr`. See this page for a discussion of `fstab`. 
You detach filesystems with the `umount` command. `umount` complains if you try to unmount a filesystem that’s in use. The filesystem to be detached must not have open files or processes whose current directories are located there, and if the filesystem contains executable programs, none of them can be running.

Linux has a “lazy” unmount option (`umount -l`) that removes a filesystem from the naming hierarchy but does not truly unmount it until all existing file references have been closed. It’s debatable whether this is a useful option. To begin with, there’s no guarantee that existing references will ever close on their own. In addition, the “semi-unmounted” state can present inconsistent filesystem semantics to the programs that are using it; they can read and write through existing file handles but cannot open new files or perform other filesystem operations.

`umount -f` force-unmounts a busy filesystem and is supported on all our example systems. However, it’s almost always a bad idea to use it on non-NFS mounts, and it may not work on certain types of filesystems (e.g., those that keep journals, such as XFS or ext4).

Instead of reaching for `umount -f` when a filesystem you’re trying to unmount turns out to be busy, run the `fuser` command to find out which processes hold references to that filesystem. `fuser -c mountpoint` prints the PID of every process that’s using a file or directory on that filesystem, plus a series of letter codes that show the nature of the activity. For example,

```
fuser -c /usr/home
/usr/home: 15897c 87787c 67124x 11201x 11199x 11198x 972x
```

The exact letter codes vary from system to system. In this example from a FreeBSD system, `c` indicates that a process has its current working directory on the filesystem and `x` indicates a program being executed. However, the details are usually unimportant—the PIDs are what you want.

To investigate the offending processes, just run `ps` with the list of PIDs returned by `fuser`. For example,

```
ps up "87787 11201"
USER  PID   %CPU %MEM  STARTED       TIME COMMAND
fnd 11201  0.0  0.2  14Jul16  2:32.49  ruby: slave_audiochannelbackend
fnd 87787  0.0  0.0  Thu07PM  0:00.93  -bash (bash)
```

Here, the quotation marks force the shell to pass the list of PIDs to `ps` as a single argument.

On Linux systems, you can avoid the need to launder PIDs through `ps` by running `fuser` with the `-v` flag. This option produces a more readable display that includes the command name.
The letter codes in the ACCESS column are the same ones used in fuser’s nonverbose output.

A more elaborate alternative to fuser is the ls/of utility. ls/of is a more complex and sophisticated program than fuser, and its output is correspondingly verbose. ls/of comes installed by default on all our example Linux systems and is available as a package on FreeBSD.

Under Linux, scripts in search of specific information about processes’ use of filesystems can also read the files in /proc directly. However, ls/of -F, which formats ls/of’s output for easy parsing, is an easier and more portable solution. Use additional command-line flags to request just the information you need.

```bash
$ fuser -cv /usr

<table>
<thead>
<tr>
<th>USER</th>
<th>PID</th>
<th>ACCESS</th>
<th>COMMAND</th>
</tr>
</thead>
<tbody>
<tr>
<td>/usr</td>
<td>root 444</td>
<td>....m</td>
<td>atd</td>
</tr>
<tr>
<td>root</td>
<td>499</td>
<td>....m</td>
<td>ssdh</td>
</tr>
<tr>
<td>root</td>
<td>520</td>
<td>....m</td>
<td>lpd</td>
</tr>
</tbody>
</table>

...
5.3 Organization of the File Tree

UNIX systems have never been well organized. Various incompatible naming conventions are used simultaneously, and different types of files are scattered randomly around the namespace. In many cases, files are divided by function and not by how likely they are to change, making it difficult to upgrade the operating system. The /etc directory, for example, contains some files that are never customized and some that are entirely local. How do you know which files to preserve during an upgrade? Well, you just have to know…or trust the installation software to make the right decisions.

As a logically minded sysadmin, you may be tempted to improve the default organization. Unfortunately, the file tree has many hidden dependencies, so such efforts usually end up creating problems. Just let everything stay where the OS installation and the system packages put it. When offered a choice of locations, always accept the default unless you have a specific and compelling reason to do otherwise.

See Chapter 11 for more information about configuring the kernel.

The root filesystem includes at least the root directory and a minimal set of files and subdirectories. The file that contains the OS kernel usually lives under /boot, but its exact name and location can vary. Under BSD and some other UNIX systems, the kernel is not really a single file so much as a set of components.

Also part of the root filesystem are /etc for critical system and configuration files, /sbin and /bin for important utilities, and sometimes /tmp for temporary files. The /dev directory was traditionally part of the root filesystem, but these days it’s a virtual filesystem that’s mounted separately. (See this page for more information.)

Some systems keep shared library files and a few other oddments, such as the C preprocessor, in the /lib or /lib64 directory. Others have moved these items into /usr/lib, sometimes leaving /lib as a symbolic link.

The directories /usr and /var are also of great importance. /usr is where most standard-but-not-system-critical programs are kept, along with various other booty such as on-line manuals and most libraries. FreeBSD stores quite a bit of local configuration under /usr/local. /var houses spool directories, log files, accounting information, and various other items that grow or change rapidly and that vary on each host. Both /usr and /var must be available to enable the system to come up all the way to multiuser mode.

See this page for some reasons why partitioning might be desirable and some rules of thumb to guide it.
In the past, it was standard practice to partition the system disk and to put some parts of the file tree on their own partitions, most commonly /usr, /var, and /tmp. That’s not uncommon even now, but the secular trend is toward having one big root filesystem. Large hard disks and increasingly sophisticated filesystem implementations have reduced the value of partitioning.

In cases where partitioning is used, it’s most frequently an attempt to prevent one part of the file tree from consuming all available space and bringing the entire system to a halt. Accordingly, /var (which contains log files that are apt to grow in times of trouble), /tmp, and user home directories are some of the most common candidates for having their own partitions. Dedicated filesystems can also store bulky items such as source code libraries and databases.

Table 5.1 lists some of the more important standard directories. (Alternate rows have been shaded to improve readability.)

Table 5.1: Standard directories and their contents
<table>
<thead>
<tr>
<th>Pathname</th>
<th>Contents</th>
</tr>
</thead>
<tbody>
<tr>
<td>/bin</td>
<td>Core operating system commands</td>
</tr>
<tr>
<td>/boot</td>
<td>Boot loader, kernel, and files needed by the kernel</td>
</tr>
<tr>
<td>/compat</td>
<td>On FreeBSD, files and libraries for Linux binary compatibility</td>
</tr>
<tr>
<td>/dev</td>
<td>Device entries for disks, printers, pseudo-terminals, etc.</td>
</tr>
<tr>
<td>/etc</td>
<td>Critical startup and configuration files</td>
</tr>
<tr>
<td>/home</td>
<td>Default home directories for users</td>
</tr>
<tr>
<td>/lib</td>
<td>Libraries, shared libraries, and commands used by /bin and /sbin</td>
</tr>
<tr>
<td>/media</td>
<td>Mount points for filesystems on removable media</td>
</tr>
<tr>
<td>/mnt</td>
<td>Temporary mount points, mounts for removable media</td>
</tr>
<tr>
<td>/opt</td>
<td>Optional software packages (rarely used, for compatibility)</td>
</tr>
<tr>
<td>/proc</td>
<td>Information about all running processes</td>
</tr>
<tr>
<td>/root</td>
<td>Home directory of the superuser (sometimes just /)</td>
</tr>
<tr>
<td>/run</td>
<td>Rendezvous points for running programs (PIDs, sockets, etc.)</td>
</tr>
<tr>
<td>/sbin</td>
<td>Core operating system commands(^a)</td>
</tr>
<tr>
<td>/srv</td>
<td>Files held for distribution through web or other servers</td>
</tr>
<tr>
<td>/sys</td>
<td>A plethora of different kernel interfaces (Linux)</td>
</tr>
<tr>
<td>/tmp</td>
<td>Temporary files that may disappear between reboots</td>
</tr>
<tr>
<td>/usr</td>
<td>Hierarchy of secondary files and commands</td>
</tr>
<tr>
<td>/usr/bin</td>
<td>Most commands and executable files</td>
</tr>
<tr>
<td>/usr/include</td>
<td>Header files for compiling C programs</td>
</tr>
<tr>
<td>/usr/lib</td>
<td>Libraries; also, support files for standard programs</td>
</tr>
<tr>
<td>/usr/local</td>
<td>Local software or configuration data; mirrors /usr</td>
</tr>
<tr>
<td>/usr/sbin</td>
<td>Less essential commands for administration and repair</td>
</tr>
<tr>
<td>/usr/share</td>
<td>Items that might be common to multiple systems</td>
</tr>
<tr>
<td>/usr/share/man</td>
<td>On-line manual pages</td>
</tr>
<tr>
<td>/usr/src</td>
<td>Source code for nonlocal software (not widely used)</td>
</tr>
<tr>
<td>/usr/tmp</td>
<td>More temporary space (preserved between reboots)</td>
</tr>
<tr>
<td>/var</td>
<td>System-specific data and a few configuration files</td>
</tr>
<tr>
<td>/var/adm</td>
<td>Varies: logs, setup records, strange administrative bits</td>
</tr>
<tr>
<td>/var/log</td>
<td>System log files</td>
</tr>
<tr>
<td>/var/run</td>
<td>Same function as /run; now often a symlink</td>
</tr>
<tr>
<td>/var/spool</td>
<td>Spooling (that is, storage) directories for printers, mail, etc.</td>
</tr>
<tr>
<td>/var/tmp</td>
<td>More temporary space (preserved between reboots)</td>
</tr>
</tbody>
</table>

\(^a\) The distinguishing characteristic of /sbin was originally that its contents were statically linked and so had fewer dependencies on other parts of the system. These days, all binaries are dynamically linked and there is no real difference between /bin and /sbin.

On most systems, a **hier** man page outlines some general guidelines for the layout of the filesystem. Don’t expect the actual system to conform to the master plan in every respect, however.

**For Linux systems, the Filesystem Hierarchy Standard** (see [wiki.linuxfoundation.org/en/FHS](http://wiki.linuxfoundation.org/en/FHS)) attempts to codify, rationalize, and explain the standard
directories. It’s an excellent resource to consult when you confront an unusual situation and need to figure out where to put something. Despite its status as a “standard,” it’s more a reflection of real-world practice than a prescriptive document. It also hasn’t undergone much updating recently, so it doesn’t describe the exact filesystem layout found on current distributions.
5.4 File Types

Most filesystem implementations define seven types of files. Even when developers add something new and wonderful to the file tree (such as the process information under /proc), it must still be made to look like one of these seven types:

- Regular files
- Directories
- Character device files
- Block device files
- Local domain sockets
- Named pipes (FIFOs)
- Symbolic links

You can determine the type of an existing file with the `file` command. Not only does `file` know about the standard types of files, but it also knows a thing or two about common formats used within regular files.

```
$ file /usr/include
/usr/include: directory
$ file /bin/sh
/bin/sh: ELF 64-bit LSB executable, x86-64, version 1 (FreeBSD),
        dynamically linked, interpreter /libexec/ld-elf.so.1, for FreeBSD
        11.0 (1100122), FreeBSD-style, stripped
```

All that hoo-hah about /bin/sh means “it’s an executable command.”

Another option for investigating files is `ls -ld`. The `-l` flag shows detailed information, and the `-d` flag forces `ls` to show the information for a directory rather than showing the directory’s contents.

The first character of the `ls` output encodes the type. For example, the circled `d` in the following output demonstrates that /usr/include is a directory:

```
$ ls -ld /usr/include
drwxr-xr-x  27 root root  4096 Jul 15 20:57 /usr/include
```

Table 5.2 shows the codes `ls` uses to represent the various types of files.

Table 5.2: File-type encoding used by `ls`
As Table 5.2 shows, **rm** is the universal tool for deleting files. But how would you delete a file named, say, `-f`? It’s a legitimate filename under most filesystems, but **rm -f** doesn’t work because **rm** interprets the `-f` as a flag. The answer is either to refer to the file by a pathname that doesn’t start with a dash (such as `./-f`) or to use **rm**’s -- argument to tell it that everything that follows is a filename and not an option (i.e., **rm** -- -f).

Filenames that contain control or Unicode characters present a similar problem since reproducing these names from the keyboard can be difficult or impossible. In this situation, you can use shell globbing (pattern matching) to identify the files to delete. When you use pattern matching, it’s a good idea to get in the habit of using **rm**’s -i option to make **rm** confirm the deletion of each file. This feature protects you against deleting any “good” files that your pattern inadvertently matches. To delete the file named `foo<Control-D>bar` in the following example, you could use

```
$ ls
foo?bar foose kde-root

$ rm -i foo
rm: remove 'foo\004bar'? y
rm: remove 'foose'? n
```

Note that **ls** shows the control character as a question mark, which can be a bit deceptive. If you don’t remember that `?` is a shell pattern-matching character and try to **rm** `foo?bar`, you might potentially remove more than one file (although not in this example). -i is your friend!

**ls -b** shows control characters as octal numbers, which can be helpful if you need to identify them specifically. `<Control-A>` is 1 (\001 in octal), `<Control-B>` is 2, and so on, in alphabetical order. **man ascii** and the Wikipedia page for ASCII both include a nice table of control characters and their octal equivalents.

To delete the most horribly named files, you might need to resort to **rm** -i *.

Another option for removing files with squirrelly names is to use an alternative interface to the filesystem such as **emacs**’s dired mode or a visual tool such as Nautilus.
**Regular files**

Regular files consist of a series of bytes; filesystems impose no structure on their contents. Text files, data files, executable programs, and shared libraries are all stored as regular files. Both sequential access and random access are allowed.
Directories

A directory contains named references to other files. You can create directories with `mkdir` and delete them with `rmdir` if they are empty. You can recursively delete nonempty directories—including all their contents—with `rm -r`.

The special entries “.” and “..” refer to the directory itself and to its parent directory; they cannot be removed. Since the root directory has no real parent directory, the path “/..” is equivalent to the path “/.” (and both are equivalent to `/`).
**Hard links**

A file’s name is stored within its parent directory, not with the file itself. In fact, more than one directory (or more than one entry in a single directory) can refer to a file at one time, and the references can have different names. Such an arrangement creates the illusion that a file exists in more than one place at the same time.

These additional references (“links,” or “hard links” to distinguish them from symbolic links, discussed below) are synonymous with the original file; as far as the filesystem is concerned, all links to the file are equivalent. The filesystem maintains a count of the number of links that point to each file and does not release the file’s data blocks until its last link has been deleted. Hard links cannot cross filesystem boundaries.

You create hard links with `ln` and remove them with `rm`. It’s easy to remember the syntax of `ln` if you keep in mind that it mirrors the syntax of `cp`. The command `cp oldfile newfile` creates a copy of `oldfile` called `newfile`, and `ln oldfile newfile` makes the name `newfile` an additional reference to `oldfile`.

In most filesystem implementations, it is technically possible to make hard links to directories as well as to flat files. However, directory links often lead to degenerate conditions such as filesystem loops and directories that don’t have a single, unambiguous parent. In most cases, a symbolic link (see [this page](#)) is a better option.

You can use `ls -l` to see how many links to a given file exist. See the `ls` example output [here](#) for some additional details. Also note the comments regarding `ls -i` on [this page](#), as this option is particularly helpful for identifying hard links.

Hard links are not a distinct type of file. Instead of defining a separate “thing” called a hard link, the filesystem simply allows more than one directory entry to point to the same file. In addition to the file’s contents, the underlying attributes of the file (such as ownerships and permissions) are also shared.
Device files let programs communicate with the system’s hardware and peripherals. The kernel includes (or loads) driver software for each of the system’s devices. This software takes care of the messy details of managing each device so that the kernel itself can remain relatively abstract and hardware-independent.

Device drivers present a standard communication interface that looks like a regular file. When the filesystem is given a request that refers to a character or block device file, it simply passes the request to the appropriate device driver. It’s important to distinguish device files from device drivers, however. The files are just rendezvous points that communicate with drivers. They are not drivers themselves.

The distinction between character and block devices is subtle and not worth reviewing in detail. In the past, a few types of hardware were represented by both block and character device files, but that configuration is rare today. As a matter of practice, FreeBSD has done away with block devices entirely, though their spectral presence can still be glimpsed in man pages and header files.

Device files are characterized by two numbers, called the major and minor device numbers. The major device number tells the kernel which driver the file refers to, and the minor device number typically tells the driver which physical unit to address. For example, major device number 4 on a Linux system denotes the serial driver. The first serial port (/dev/tty0) would have major device number 4 and minor device number 0.

Drivers can interpret the minor device numbers that are passed to them in whatever way they please. For example, tape drivers use the minor device number to determine whether the tape should be rewound when the device file is closed.

In the distant past, /dev was a generic directory and the device files within it were created with mknod and removed with rm. Unfortunately, this crude system was ill-equipped to deal with the endless sea of drivers and device types that have appeared over the last few decades. It also facilitated all sorts of potential configuration mismatches: device files that referred to no actual device, devices inaccessible because they had no device files, and so on.

These days, the /dev directory is normally mounted as a special filesystem type, and its contents are automatically maintained by the kernel in concert with a user-level daemon. There are a couple of different versions of this same basic system. See Chapter 11, Drivers and the Kernel, for more information about each system’s approach to this task.
Local domain sockets

Sockets are connections between processes that allow them to communicate hygienically. UNIX defines several kinds of sockets, most of which involve the network.

See Chapter 10 for more information about syslog.

Local domain sockets are accessible only from the local host and are referred to through a filesystem object rather than a network port. They are sometimes known as “UNIX domain sockets.” Syslog and the X Window System are examples of standard facilities that use local domain sockets, but there are many more, including many databases and app servers.

Local domain sockets are created with the `socket` system call and removed with the `rm` command or the `unlink` system call once they have no more users.
Named pipes
Like local domain sockets, named pipes allow communication between two processes running on the same host. They’re also known as “FIFO files” (As in financial accounting, FIFO is short for the phrase “first in, first out”). You can create named pipes with `mknod` and remove them with `rm`.

Named pipes and local domain sockets serve similar purposes, and the fact that both exist is essentially a historical artifact. Most likely, neither of them would exist if UNIX and Linux were designed today; network sockets would stand in for both.
Symbolic links

A symbolic or “soft” link points to a file by name. When the kernel comes upon a symbolic link in the course of looking up a pathname, it redirects its attention to the pathname stored as the contents of the link. The difference between hard links and symbolic links is that a hard link is a direct reference, whereas a symbolic link is a reference by name. Symbolic links are distinct from the files they point to.

You create symbolic links with `ln -s` and remove them with `rm`. Since symbolic links can contain arbitrary paths, they can refer to files on other filesystems or to nonexistent files. A series of symbolic links can also form a loop.

A symbolic link can contain either an absolute or a relative path. For example,

```bash
$ sudo ln -s archived/secure /var/data/secure
```

links `/var/data/secure` to `/var/data/archived/secure` with a relative path. It creates the symbolic link `/var/data/secure` with a target of `archived/secure`, as demonstrated by this output from `ls`:

```bash
$ ls -l /var/data/secure
lrwxrwxrwx 1 root root 18 Aug 3 12:54 /var/data/secure -> archived/secure
```

The entire `/var/data` directory could then be moved elsewhere without causing the symbolic link to stop working.

The file permissions that `ls` shows for a symbolic link, `lrwxrwxrwx`, are dummy values. Permission to create, remove, or follow the link is controlled by the containing directory, whereas read, write, and execute permission on the link target are granted by the target’s own permissions. Therefore, symbolic links do not need (and do not have) any permission information of their own.

A common mistake is to think that the first argument to `ln -s` is interpreted relative to the current working directory. However, that argument is not actually resolved as a filename by `ln`: it’s simply a literal string that becomes the target of the symbolic link.
5.5 **File attributes**

Under the traditional UNIX and Linux filesystem model, every file has a set of nine permission bits that control who can read, write, and execute the contents of the file. Together with three other bits that primarily affect the operation of executable programs, these bits constitute the file’s “mode.”

The twelve mode bits are stored along with four bits of file-type information. The four file-type bits are set when the file is first created and cannot be changed, but the file’s owner and the superuser can modify the twelve mode bits with the **chmod** (change mode) command. Use `ls -l` (or `ls -ld` for a directory) to inspect the values of these bits. See this page for an example.
The permission bits

Nine permission bits determine what operations can be performed on a file and by whom. Traditional UNIX does not allow permissions to be set per user (although all systems now support access control lists of one sort or another; see this page). Instead, three sets of permissions define access for the owner of the file, the group owners of the file, and everyone else (in that order). Each set has three bits: a read bit, a write bit, and an execute bit (also in that order).

If you think of the owner as “the user” and everyone else as “other,” you can remember the order of the permission sets by thinking of the name Hugo. u, g, and o are also the letter codes used by the mnemonic version of chmod.

It’s convenient to discuss file permissions in terms of octal (base 8) numbers because each digit of an octal number represents three bits and each group of permission bits consists of three bits. The topmost three bits (with octal values of 400, 200, and 100) control access for the owner. The second three (40, 20, and 10) control access for the group. The last three (4, 2, and 1) control access for everyone else (“the world”). In each triplet, the high bit is the read bit, the middle bit is the write bit, and the low bit is the execute bit.

Although a user might fit into two of the three permission categories, only the most specific permissions apply. For example, the owner of a file always has access determined by the owner permission bits and never by the group permission bits. It is possible for the “other” and “group” categories to have more access than the owner, although this configuration would be highly unusual.

On a regular file, the read bit allows the file to be opened and read. The write bit allows the contents of the file to be modified or truncated; however, the ability to delete or rename (or delete and then re-create!) the file is controlled by the permissions on its parent directory, where the name-to-dataspace mapping is actually stored.

The execute bit allows the file to be executed. Two types of executable files exist: binaries, which the CPU runs directly, and scripts, which must be interpreted by a shell or some other program. By convention, scripts begin with a line similar to

```
#!/usr/bin/perl
```

that specifies an appropriate interpreter. Nonbinary executable files that do not specify an interpreter are assumed to be sh scripts.

The kernel understands the #! ("shebang") syntax and acts on it directly. However, if the interpreter is not specified completely and correctly, the kernel will refuse to execute the file. The shell then makes a second attempt to execute the script by calling /bin/sh, which is usually a link to the Almquist shell or to bash; see this page. Sven Mascheck maintains an excruciatingly
detailed page about the history, implementation, and cross-platform behavior of the shebang at goo.gl/J7izhL.

For a directory, the execute bit (often called the “search” or “scan” bit in this context) allows the directory to be entered or passed through as a pathname is evaluated, but not to have its contents listed. The combination of read and execute bits allows the contents of the directory to be listed. The combination of write and execute bits allows files to be created, deleted, and renamed within the directory.

A variety of extensions such as access control lists (see this page), SELinux (see this page), and “bonus” permission bits defined by individual filesystems (see this page) complicate or override the traditional 9-bit permission model. If you’re having trouble explaining the system’s observed behavior, check to see whether one of these factors might be interfering.
The setuid and setgid bits

The bits with octal values 4000 and 2000 are the setuid and setgid bits. When set on executable files, these bits allow programs to access files and processes that would otherwise be off-limits to the user that runs them. The setuid/setgid mechanism for executables is described here.

When set on a directory, the setgid bit causes newly created files within the directory to take on the group ownership of the directory rather than the default group of the user that created the file. This convention makes it easier to share a directory of files among several users, as long as they belong to a common group. This interpretation of the setgid bit is unrelated to its meaning when set on an executable file, but no ambiguity can exist as to which meaning is appropriate.
The sticky bit

The bit with octal value 1000 is called the sticky bit. It was of historical importance as a modifier for executable files on early UNIX systems. However, that meaning of the sticky bit is now obsolete and modern systems silently ignore the sticky bit when it’s set on regular files.

If the sticky bit is set on a directory, the filesystem won’t allow you to delete or rename a file unless you are the owner of the directory, the owner of the file, or the superuser. Having write permission on the directory is not enough. This convention helps make directories like /tmp a little more private and secure.
**ls: list and inspect files**

The filesystem maintains about forty separate pieces of information for each file, but most of them are useful only to the filesystem itself. As a system administrator, you will be concerned mostly with the link count, owner, group, mode, size, last access time, last modification time, and type. You can inspect all these with `ls -l` (or `ls -ld` for a directory; without the `-d` flag, `ls` lists the directory’s contents).

An attribute change time is also maintained for each file. The conventional name for this time (the “ctime,” short for “change time”) leads some people to believe that it is the file’s creation time. Unfortunately, it is not; it just records the time at which the attributes of the file (owner, mode, etc.) were last changed (as opposed to the time at which the file’s contents were modified).

Consider the following example:

```
$ ls -l /usr/bin/gzip
-rwxr-xr-x  4 root wheel  37432 Nov 11 2016 /usr/bin/gzip
```

The first field specifies the file’s type and mode. The first character is a dash, so the file is a regular file. (See Table 5.2 for other codes.)

The next nine characters in this field are the three sets of permission bits. The order is owner-group-other, and the order of bits within each set is read-write-execute. Although these bits have only binary values, `ls` shows them symbolically with the letters `r`, `w`, and `x` for read, write, and execute. In this case, the owner has all permissions on the file and everyone else has read and execute permission.

If the setuid bit had been set, the `x` representing the owner’s execute permission would have been replaced with an `s`, and if the setgid bit had been set, the `x` for the group would also have been replaced with an `s`. The last character of the permissions (execute permission for “other”) is shown as `t` if the sticky bit of the file is turned on. If either the setuid/setgid bit or the sticky bit is set but the corresponding execute bit is not, these bits are shown as `S` or `T`.

The next field in the listing is the file’s link count. In this case it is 4, indicating that `/usr/bin/gzip` is just one of four names for this file (the others on this system are `gunzip`, `gzcat`, and `zcat`, all in `/usr/bin`). Each time a hard link is made to a file, the file’s link count is incremented by 1. Symbolic links do not affect the link count.

All directories have at least two hard links: the link from the parent directory and the link from the special file called `.` inside the directory itself.

The next two fields in the `ls` output are the owner and group owner of the file. In this example, the file’s owner is root, and the file belongs to the group named wheel. The filesystem actually stores these as the user and group ID numbers rather than as names. If the text versions (names) can’t be determined, `ls` shows the fields as numbers. This might happen if the user or group that
owns the file has been deleted from the /etc/passwd or /etc/group file. It could also suggest a problem with your LDAP database (if you use one); see Chapter 17.

The next field is the size of the file in bytes. This file is 37,432 bytes long. Next comes the date of last modification: November 11, 2016. The last field in the listing is the name of the file, /usr/bin/gzip.

ls output is slightly different for a device file. For example:

```
$ ls -l /dev/tty0
crw------- 1 root tty 4, 0 Aug 3 15:12 /dev/tty0
```

Most fields are the same, but instead of a size in bytes, ls shows the major and minor device numbers. /dev/tty0 is the first virtual console on this (Red Hat) system and is controlled by device driver 4 (the terminal driver). The dot at the end of the mode indicates the absence of an access control list (ACL, discussed starting here). Some systems show this by default and some don’t.

One ls option that’s useful for scoping out hard links is -i, which tells ls to show each file’s “inode number.” Briefly, the inode number is an integer associated with the contents of a file. Inodes are the “things” that are pointed to by directory entries; entries that are hard links to the same file have the same inode number. To figure out a complex web of links, you need both ls -li, to show link counts and inode numbers, and find, to search for matches. (Try find mountpoint -xdev -inum inode -print.)

Some other ls options that are important to know are -a to show all entries in a directory (even files whose names start with a dot), -t to sort files by modification time (or -tr to sort in reverse chronological order), -F to show the names of files in a way that distinguishes directories and executable files, -R to list recursively, and -h to show file sizes in human-readable form (e.g., 8K or 53M).

Most versions of ls now default to color-coding files if your terminal program supports this (most do). ls specifies colors according to a limited and abstract palette (“red,” “blue,” etc.), and it’s up to the terminal program to map these requests to specific colors. You may need to tweak both ls (the LSCOLORS or LS_COLORS environment variable) and the terminal emulator to achieve colors that are readable and unobtrusive. Alternatively, you can just remove the default configuration for colorization (usually /etc/profile.d/colorls*) to eliminate colors entirely.
**chmod: change permissions**

The `chmod` command changes the permissions on a file. Only the owner of the file and the superuser can change a file’s permissions. To use the command on early UNIX systems, you had to learn a bit of octal notation, but current versions accept both octal notation and a mnemonic syntax. The octal syntax is generally more convenient for administrators, but it can only be used to specify an absolute value for the permission bits. The mnemonic syntax can modify some bits but leave others alone.

The first argument to `chmod` is a specification of the permissions to be assigned, and the second and subsequent arguments are names of files on which permissions should be changed. In the octal case, the first octal digit of the specification is for the owner, the second is for the group, and the third is for everyone else. If you want to turn on the setuid, setgid, or sticky bits, you use four octal digits rather than three, with the three special bits forming the first digit.

Table 5.3 illustrates the eight possible combinations for each set of three bits, where `r`, `w`, and `x` stand for read, write, and execute.

<table>
<thead>
<tr>
<th>Octal</th>
<th>Binary</th>
<th>Perms</th>
<th>Octal</th>
<th>Binary</th>
<th>Perms</th>
</tr>
</thead>
<tbody>
<tr>
<td>0 000</td>
<td>---</td>
<td></td>
<td>4 100</td>
<td>r--</td>
<td></td>
</tr>
<tr>
<td>1 001</td>
<td>--x</td>
<td></td>
<td>5 101</td>
<td>r-x</td>
<td></td>
</tr>
<tr>
<td>2 010</td>
<td>-w-</td>
<td></td>
<td>6 110</td>
<td>r-w-</td>
<td></td>
</tr>
<tr>
<td>3 011</td>
<td>-wx</td>
<td></td>
<td>7 111</td>
<td>rwx</td>
<td></td>
</tr>
</tbody>
</table>

For example, `chmod 711 myprog` gives all permissions to the user (owner) and execute-only permission to everyone else. (If `myprog` were a shell script, it would need both read and execute permission turned on. For the script to be run by an interpreter, it must be opened and read like a text file. Binary files are executed directly by the kernel and therefore do not need read permission turned on.)

For the mnemonic syntax, you combine a set of targets (`u`, `g`, or `o` for user, group, other, or `a` for all three) with an operator (`+`, `-`, `=` to add, remove, or set) and a set of permissions. The `chmod` man page gives the details, but the syntax is probably best learned by example. Table 5.4 exemplifies some mnemonic operations.

**Table 5.4: Examples of chmod’s mnemonic syntax**
The hard part about using the mnemonic syntax is remembering whether `o` stands for “owner” or “other”; “other” is correct. Just remember `u` and `g` by analogy to UID and GID; only one possibility is left. Or remember the order of letters in the name Hugo.

On Linux systems, you can also specify the modes to be assigned by copying them from an existing file. For example, `chmod --reference=filea fileb` makes `fileb`’s mode the same as `filea`’s.

With the `-R` option, `chmod` recursively updates the file permissions within a directory. However, this feat is trickier than it looks because the enclosed files and directories may not share the same attributes; for example, some might be executable files; others, text files. Mnemonic syntax is particularly useful with `-R` because it preserves bits whose values you don’t set explicitly. For example,

```bash
$ chmod -R g+w mydir
```

adds group write permission to `mydir` and all its contents without messing up the execute bits of directories and programs.

If you want to adjust execute bits, be wary of `chmod -R`. It’s blind to the fact that the execute bit has a different interpretation on a directory than it does on a flat file. Therefore, `chmod -R a-x` probably won’t do what you intend. Use `find` to select only the regular files:

```bash
$ find mydir -type f -exec chmod a-x {} ';'
```
**chown and chgrp: change ownership and group**

The *chown* command changes a file’s ownership, and the *chgrp* command changes its group ownership. The syntax of *chown* and *chgrp* mirrors that of *chmod*, except that the first argument is the new owner or group, respectively.

To change a file’s group, you must either be the superuser or be the owner of the file and belong to the group you’re changing to. Older systems in the SysV lineage allowed users to give away their own files with *chown*, but that’s unusual these days; *chown* is now a privileged operation.

Like *chmod*, *chown* and *chgrp* offer the recursive -R flag to change the settings of a directory and all the files underneath it. For example, the sequence

```bash
$ sudo chown -R matt ~/matt/restore
$ sudo chgrp -R staff ~/matt/restore
```

could reset the owner and group of files restored from a backup for the user matt. Don’t try to *chown* dot files with a command such as

```bash
$ sudo chown -R matt ~/matt/.*
```

since the pattern matches `~/matt/..` and therefore ends up changing the ownerships of the parent directory and probably the home directories of other users.

*chown* can change both the owner and group of a file at once with the syntax

```
chown user:group file ...
```

For example,

```bash
$ sudo chown -R matt:staff ~/matt/restore
```

You can actually omit either *user* or *group*, which makes the *chgrp* command superfluous. If you include the colon but name no specific *group*, the Linux version of *chown* uses the user’s default group.

Some systems accept the notation *user.group* as being equivalent to *user:group*. This is just a nod to historical variation among systems; it means the same thing.
umask: assign default permissions

You can use the built-in shell command **umask** to influence the default permissions given to the files you create. Every process has its own **umask** attribute; the shell’s built-in **umask** command sets the shell’s own **umask**, which is then inherited by commands that you run.

The **umask** is specified as a three-digit octal value that represents the permissions to *take away*. When a file is created, its permissions are set to whatever the creating program requests minus whatever the **umask** forbids. Thus, the individual digits of the **umask** allow the permissions shown in Table 5.5.

### Table 5.5: Permission encoding for umask

<table>
<thead>
<tr>
<th>Octal</th>
<th>Binary</th>
<th>Perms</th>
<th>Octal</th>
<th>Binary</th>
<th>Perms</th>
</tr>
</thead>
<tbody>
<tr>
<td>0</td>
<td>000</td>
<td>rwx</td>
<td>4</td>
<td>100</td>
<td>-wx</td>
</tr>
<tr>
<td>1</td>
<td>001</td>
<td>r-w</td>
<td>5</td>
<td>101</td>
<td>-w</td>
</tr>
<tr>
<td>2</td>
<td>010</td>
<td>r-x</td>
<td>6</td>
<td>110</td>
<td>--x</td>
</tr>
<tr>
<td>3</td>
<td>011</td>
<td>r--</td>
<td>7</td>
<td>111</td>
<td>---</td>
</tr>
</tbody>
</table>

For example, **umask** 027 allows all permissions for the owner but forbids write permission to the group and allows no permissions for anyone else. The default **umask** value is often 022, which denies write permission to the group and world but allows read permission.

---

See **Chapter 8** for more information about startup files.

---

In the standard access control model, you cannot force users to have a particular **umask** value because they can always reset it to whatever they want. However, you can put a suitable default in the sample startup files that you give to new users. If you require more control over the permissions on user-created files, you’ll need to graduate to a mandatory access control system such as SELinux; see this page.
Linux bonus flags

Linux defines a set of supplemental flags that can be set on files to request special handling. For example, the `a` flag makes a file append-only, and the `i` flag makes it immutable and undeletable.

Flags have binary values, so they are either present or absent for a given file. The underlying filesystem implementation must support the corresponding feature, so not all flags can be used on all filesystem types. In addition, some flags are experimental, unimplemented, or read-only.

Linux uses the commands `lsattr` and `chattr` to view and change file attributes. Table 5.6 lists some of the more mainstream flags.

<table>
<thead>
<tr>
<th>Flag</th>
<th>FS</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>A</td>
<td>XBE</td>
<td>Never update access time (<code>st_atime</code>; for performance)</td>
</tr>
<tr>
<td>a</td>
<td>XBE</td>
<td>Allow writing only in append mode(^b)</td>
</tr>
<tr>
<td>C</td>
<td>B</td>
<td>Disable copy-on-write updates</td>
</tr>
<tr>
<td>c</td>
<td>B</td>
<td>Compress contents</td>
</tr>
<tr>
<td>D</td>
<td>BE</td>
<td>Force directory updates to be written synchronously</td>
</tr>
<tr>
<td>d</td>
<td>XBE</td>
<td>Do not back up; backup utilities should ignore this file</td>
</tr>
<tr>
<td>i</td>
<td>XBE</td>
<td>Make file immutable and undeletable(^b)</td>
</tr>
<tr>
<td>j</td>
<td>E</td>
<td>Keep a journal for data changes as well as metadata</td>
</tr>
<tr>
<td>S</td>
<td>XBE</td>
<td>Force changes to be written synchronously (no buffering)</td>
</tr>
<tr>
<td>X</td>
<td>B</td>
<td>Avoid data compression if it is the default</td>
</tr>
</tbody>
</table>

a. X = XFS, B = Btrfs, E = ext3 and ext4  
b. Can be set only by root

As might be expected from such a random grab bag of features, the value of these flags to administrators varies. The main thing to remember is that if a particular file seems to be behaving strangely, check it with `lsattr` to see if it has one or more flags enabled.

Waiving maintenance of last-access times (the `A` flag) can boost performance in some situations. However, its value depends on the filesystem implementation and access pattern; you’ll have to do your own benchmarking. In addition, modern kernels now default to mounting filesystems with the `relatime` option, which minimizes updates to `st_atime` and makes the `A` flag largely obsolete.

The immutable and append-only flags (`i` and `a`) were largely conceived as ways to make the system more resistant to tampering by hackers or hostile code. Unfortunately, they can confuse software and protect only against hackers that don’t know enough to use `chattr -ia`. Real-world experience has shown that these flags are more often used by hackers than against them.

See Chapter 5 for more information about configuration management.
We have seen several cases in which admins have used the i (immutable) flag to prevent changes that would otherwise be imposed by a configuration management system such as Ansible or Salt. Needless to say, this hack creates confusion once the details have been forgotten and no one can figure out why configuration management isn’t working. Never do this—just think of the shame your mother would feel if she knew what you’d been up to. Fix the issue within the configuration management system like Mom would want.

The “no backup” flag (d) is potentially of interest to administrators, but since it’s an advisory flag, make sure that your backup system honors it.

Flags that affect journaling and write synchrony (D, j, and S) exist primarily to support databases. They are not of general use for administrators. All these options can reduce filesystem performance significantly. In addition, tampering with write synchrony has been known to confuse fsck on ext* filesystems.
5.6 Access control lists

The traditional 9-bit owner/group/other access control system is powerful enough to accommodate the vast majority of administrative needs. Although the system has clear limitations, it’s very much in keeping with the UNIX traditions (some might say, “former traditions”) of simplicity and predictability.

Access control lists, aka ACLs, are a more powerful but also more complicated way of regulating access to files. Each file or directory can have an associated ACL that lists the permission rules to be applied to it. Each of the rules within an ACL is called an access control entry or ACE.

An access control entry identifies the user or group to which it applies and specifies a set of permissions to be applied to those entities. ACLs have no set length and can include permission specifications for multiple users or groups. Most OSes limit the length of an individual ACL, but the limit is high enough (usually at least 32 entries) that it rarely comes into play.

The more sophisticated ACL systems let administrators specify partial sets of permissions or negative permissions. Most also have inheritance features that allow access specifications to propagate to newly created filesystem entities.
A cautionary note

ACLs are widely supported and occupy our attention for the rest of this chapter. However, neither of these facts should be interpreted as an encouragement to embrace them. ACLs have a niche, but it lies outside the mainstream of UNIX and Linux administration.

ACLs exist primarily to facilitate Windows compatibility and to serve the needs of the small segment of enterprises that actually require ACL-level flexibility. They are not the shiny next generation of access control and are not intended to supplant the traditional model.

ACLs’ complexity creates several potential problems. Not only are ACLs tedious to use, but they can also cause unexpected interactions with ACL-unaware backup systems, network file service peers, and even simple programs such as text editors.

ACLs also tend to become increasingly unmaintainable as the number of entries grows. Real-world ACLs frequently include vestigial entries and entries that serve only to compensate for issues caused by previous entries. It’s possible to refactor and simplify these complex ACLs, but that’s risky and time consuming, so it rarely gets done.

In the past, copies of this chapter that we’ve sent out to professional administrators for review have often come back with notes such as, “This part looks fine, but I can’t really say, because I’ve never used ACLs.”
ACL types

Two types of ACLs have emerged as the predominant standards for UNIX and Linux: POSIX ACLs and NFSv4 ACLs.

The POSIX version dates back to specification work done in the mid-1990s. Unfortunately, no actual standard was ever issued, and initial implementations varied widely. These days, we are in much better shape. Systems have largely converged on a common framing for POSIX ACLs and a common command set, `getfacl` and `setfacl`, for manipulating them.

To a first approximation, the POSIX ACL model simply extends the traditional UNIX `rwx` permission system to accommodate permissions for multiple groups and users.

As POSIX ACLs were coming into focus, it became increasingly common for UNIX and Linux to share filesystems with Windows, which has its own set of ACL conventions. Here the plot thickens, because Windows makes a variety of distinctions that are not found in either the traditional UNIX model or its POSIX ACL equivalent. Windows ACLs are semantically more complex, too; for example, they allow negative permissions (“deny” entries) and have a complicated inheritance scheme.

See Chapter 21 for more information about NFS.

The architects of version 4 of NFS—a common file-sharing protocol—wanted to incorporate ACLs as a first-class entity. Because of the UNIX/Windows split and the inconsistencies among UNIX ACL implementations, it was clear that the systems on the ends of an NFSv4 connection might often be of different types. Each system might understand NFSv4 ACLs, POSIX ACLs, Windows ACLs, or no ACLs at all. The NFSv4 standard would have to be interoperable with all these various worlds without causing too many surprises or security problems.

Given this constraint, it’s perhaps not surprising that NFSv4 ACLs are essentially a union of all preexisting systems. They are a strict superset of POSIX ACLs, so any POSIX ACL can be represented as an NFSv4 ACL without loss of information. At the same time, NFSv4 ACLs accommodate all the permission bits found on Windows systems, and they have most of Windows’ semantic features as well.
Implementation of ACLs

In theory, responsibility for maintaining and enforcing ACLs could be assigned to several different components of the operating system. ACLs could be implemented by the kernel on behalf of all the system’s filesystems, by individual filesystems, or perhaps by higher-level software such as NFS and SMB servers.

In practice, ACL support is both OS-dependent and filesystem-dependent. A filesystem that supports ACLs on one system might not support them on another, or it might feature a somewhat different implementation managed by different commands.

See Chapter 22 for more information about SMB.

File service daemons map their host’s native ACL scheme (or schemes) to and from the conventions appropriate to the filing protocol: NFSv4 ACLs for NFS, and Windows ACLs for SMB. The details of that mapping depend on the implementation of the file server. Usually, the rules are complicated and somewhat tunable with configuration options.

Because ACL implementations are filesystem-specific and because systems support multiple filesystem implementations, some systems end up supporting multiple types of ACLs. Even a given filesystem might offer several ACL options, as seen in the various ports of ZFS. If multiple ACL systems are available, the commands to manipulate them might be the same or different; it depends on the system. Welcome to sysadmin hell.
Linux ACL support

Linux has standardized on POSIX-style ACLs. NFSv4 ACLs are not supported at the filesystem level, though of course Linux systems can mount and share NFSv4 filesystems over the network.

An advantage of this standardization is that nearly all Linux filesystems now include POSIX ACL support, including XFS, Btrfs, and the ext* family. Even ZFS, whose native ACL system is NFSv4-ish, has been ported to Linux with POSIX ACLs. The standard `getfacl` and `setfacl` commands can be used everywhere, without regard to the underlying filesystem type. (You may, however, need to ensure that the correct `mount` option has been used to mount the filesystem. Filesystems generally support an `acl` option, a `noacl` option, or both, depending on their defaults.)

Linux does have a command suite (`nfs4_getfacl`, `nfs4_setfacl`, and `nfs4_editfacl`) for grooming the NFSv4 ACLs of files mounted from NFS servers. However, these commands cannot be used on locally stored files. Moreover, they are rarely included in distributions’ default software inventory; you’ll have to install them separately.
FreeBSD ACL support

FreeBSD supports both POSIX ACLs and NFSv4 ACLs. Its native `getfacl` and `setfacl` commands have been extended to include NFSv4-style ACL wrangling. NSFv4 ACL support is a relatively recent (as of 2017) development.

At the filesystem level, both UFS and ZFS support NFSv4-style ACLs, and UFS supports POSIX ACLs as well. The potential point of confusion here is ZFS, which is NFSv4-only on BSD (and on Solaris, its system of origin) and POSIX-only on Linux.

For UFS, use one of the `mount` options `acls` or `nfsv4acls` to specify which world you want to live in. These options are mutually exclusive.
POSIX ACLs

POSIX ACLs are a mostly straightforward extension of the standard 9-bit UNIX permission model. Read, write, and execute permission are the only capabilities that the ACL system deals with. Embellishments such as the setuid and sticky bits are handled exclusively through the traditional mode bits.

ACLs allow the rwx bits to be set independently for any combination of users and groups. Table 5.7 shows what the individual entries in an ACL can look like.

**Table 5.7: Entries that can appear in POSIX ACLs**

<table>
<thead>
<tr>
<th>Format</th>
<th>Example</th>
<th>Sets permissions for</th>
</tr>
</thead>
<tbody>
<tr>
<td>user::perms</td>
<td>user::rwx-</td>
<td>The file’s owner</td>
</tr>
<tr>
<td>user:username:perms</td>
<td>user:trent:rw-</td>
<td>A specific user</td>
</tr>
<tr>
<td>group::perms</td>
<td>group::r-x</td>
<td>The group that owns the file</td>
</tr>
<tr>
<td>group:groupname:perms</td>
<td>group:staff:rw-</td>
<td>A specific group</td>
</tr>
<tr>
<td>other::perms</td>
<td>other::-</td>
<td>All others</td>
</tr>
<tr>
<td>mask::perms</td>
<td>mask::rwx</td>
<td>All but owner and other(^a)</td>
</tr>
</tbody>
</table>

\(^a\) Masks are somewhat tricky and are explained later in this section.

Users and groups can be identified by name or by UID/GID. The exact number of entries that an ACL can contain varies with the filesystem implementation but is usually at least 32. That’s probably about the practical limit for manageability, anyway.

**Interaction between traditional modes and ACLs**

Files with ACLs retain their original mode bits, but consistency is automatically enforced and the two sets of permissions can never conflict. The following example demonstrates that the ACL entries automatically update in response to changes made with the standard `chmod` command:
$ touch example
$ ls -l example
-rw-rw-r-- 1 garth garth 0 Jun 14 15:57 example
$ getfacl example
# file: example
# owner: garth
# group: garth
user::rw-
group::rw-
other::r--
$ chmod 640 example
$ ls -l example
-rw-r----- 1 garth garth 0 Jun 14 15:57 example
$ getfacl --omit-header example
user::rw-
group::r--
other::---

(This example is from Linux. The FreeBSD version of `getfacl` uses `-q` instead of `--omit-header` to suppress the comment-like lines in the output.)

This enforced consistency allows older software with no awareness of ACLs to play reasonably well in the ACL world. However, there’s a twist. Even though the `group::` ACL entry in the example above appears to be tracking the middle set of traditional mode bits, that will not always be the case.

To understand why, suppose that a legacy program clears the write bits within all three permission sets of the traditional mode (e.g., `chmod ugo-w file`). The intention is clearly to make the file unwritable by anyone. But what if the resulting ACL were to look like this?

user::r--
group::r--
group:staff:rw-
other::r--

From the perspective of legacy programs, the file appears to be unmodifiable, yet it is actually writable by anyone in group staff. Not good. To reduce the chance of ambiguity and misunderstandings, the following rules are enforced:

- The `user::` and `other::` ACL entries are by definition identical to the “owner” and “other” permission bits from the traditional mode. Changing the mode changes the corresponding ACL entries, and vice versa.

- In all cases, the effective access permission afforded to the file’s owner and to users not mentioned in another way are those specified in the `user::` and `other::` ACL entries, respectively.
• If a file has no explicitly defined ACL or has an ACL that consists of only one user:, one group:, and one other: entry, these ACL entries are identical to the three sets of traditional permission bits. This is the case illustrated in the `getfacl` example above. (Such an ACL is termed “minimal” and need not actually be implemented as a logically separate ACL.)

• In more complex ACLs, the traditional group permission bits correspond to a special ACL entry called mask rather than the group:: ACL entry. The mask limits the access that the ACL can confer on all named users, all named groups, and the default group.

In other words, the mask specifies an upper bound on the access that the ACL can assign to individual groups and users. It is conceptually similar to the `umask`, except that the ACL mask is always in effect and that it specifies the allowed permissions rather than the permissions to be denied. ACL entries for named users, named groups, and the default group can include permission bits that are not present in the mask, but filesystems simply ignore them.

As a result, the traditional mode bits can never understate the access allowed by the ACL as a whole. Furthermore, clearing a bit from the group portion of the traditional mode clears the corresponding bit in the ACL mask and thereby forbids this permission to everyone but the file’s owner and those who fall in the category of “other.”

When the ACL shown in the previous example is expanded to include entries for a specific user and group, `setfacl` automatically supplies an appropriate mask:

```bash
$ ls -l example
-rw-r----- 1 garth garth 0 Jun 14 15:57 example
$ setfacl -m user::r,user:trent:rw,group:admin:rw example
$ ls -l example
-r--rw----- 1 garth garth 0 Jun 14 15:57 example
$ getfacl --omit-header example
user::r--
user:trent:rw-
group::r--
group:admin:rw-
mask::rw-
other::---
```

The `-m` option to `setfacl` means “modify”: it adds entries that are not already present and adjusts those that are already there. Note that `setfacl` automatically generates a mask that allows all the permissions granted in the ACL to take effect. If you want to set the mask by hand, include it in the ACL entry list given to `setfacl` or use the `-n` option to prevent `setfacl` from regenerating it.

Note that after the `setfacl` command, `ls -l` shows a `+` sign at the end of the file’s mode to denote that it now has a real ACL associated with it. The first `ls -l` shows no `+` because at that point the ACL is “minimal.”
If you use the traditional `chmod` command to manipulate an ACL-bearing file, be aware that your settings for the “group” permissions affect only the mask. To continue the previous example:

```
$ chmod 770 example
$ ls -l example
-rwxrwx---- 1 garth staff 0 Jun 14 15:57 example
$ getfacl --omit-header example
user::rwx
user:trent:rw-
group::r--
group:admin:rw-
mask::rwx
other::---
```

The `ls` output in this case is misleading. Despite the apparently generous group permissions, no one actually has permission to execute the file by reason of group membership. To grant such permission, you must edit the ACL itself.

To remove an ACL entirely and revert to the standard UNIX permission system, use `setfacl -bn`. (Strictly speaking, the `-n` flag is needed only on FreeBSD. Without it, FreeBSD’s `setfacl` leaves you with a vestigial mask entry that will screw up later group-mode changes. However, you can include the `-n` on Linux without creating problems.)

**POSIX access determination**

When a process attempts to access a file, its effective UID is compared to the UID that owns the file. If they are the same, access is determined by the ACL’s `user::` permissions. Otherwise, if a matching user-specific ACL entry exists, permissions are determined by that entry in combination with the ACL mask.

If no user-specific entry is available, the filesystem tries to locate a valid group-related entry that authorizes the requested access; these entries are processed in conjunction with the ACL mask. If no matching entry can be found, the `other::` entry prevails.

**POSIX ACL inheritance**

In addition to the ACL entry types listed in Table 5.7, the ACLs for directories can include default entries that are propagated to the ACLs of newly created files and subdirectories created within them. Subdirectories receive these entries both in the form of active ACL entries and in the form of copies of the default entries. Therefore, the original default entries may eventually propagate down through several layers of the directory hierarchy.

Once default entries have been copied to new subdirectories, there is no ongoing connection between the parent and child ACLs. If the parent’s default entries change, those changes are not reflected in the ACLs of existing subdirectories.
You can set default ACL entries with `setfacl -dm`. Alternatively, you can include default entries within a regular access control entry list by prefixing them with `default:`.

If a directory has *any* default entries, it must include a full set of defaults for `user::`, `group::`, `other::`, and `mask::`. `setfacl` will fill in any default entries you don’t specify by copying them from the current permissions ACL, generating a summary mask as usual.
NFSv4 ACLs

In this section, we discuss the characteristics of NFSv4 ACLs and briefly review the command syntax used to set and inspect them on FreeBSD. They aren’t supported on Linux (other than by NFS service daemons).

From a structural perspective, NFSv4 ACLs are similar to Windows ACLs. The main difference between them lies in the specification of the entity to which an access control entry refers.

In both systems, the ACL stores this entity as a string. For Windows ACLs, the string typically contains a Windows security identifier (SID), whereas for NFSv4, the string is typically of the form user:username or group:groupname. It can also be one of the special tokens owner@, group@, or everyone@. These latter entries are the most common because they correspond to the mode bits found on every file.

Systems such as Samba that share files between UNIX and Windows systems must provide some way of mapping between Windows and NFSv4 identities.

The NFSv4 and Windows permission models are more granular than the traditional UNIX read-write-execute model. In the case of NFSv4, the main refinements are as follows:

- NFSv4 distinguishes permission to create files within a directory from permission to create subdirectories.
- NFSv4 has a separate “append” permission bit.
- NFSv4 has separate read and write permissions for data, file attributes, extended attributes, and ACLs.
- NFSv4 controls a user’s ability to change the ownership of a file through the standard ACL system. In traditional UNIX, the ability to change the ownership of files is usually reserved for root.

Table 5.8 shows the various permissions that can be assigned in the NFSv4 system. It also shows the one-letter codes used to represent them and the more verbose canonical names.

Table 5.8: NFSv4 file permissions
Although the NFSv4 permission model is fairly detailed, the individual permissions should mostly be self-explanatory. (The “synchronize” permission allows a client to specify that its modifications to a file should be synchronous—that is, calls to write should not return until the data has actually been saved on disk.)

An extended attribute is a named chunk of data that is stored along with a file; most modern filesystems support such attributes. At this point, the predominant use of extended attributes is to store ACLs themselves. However, the NFSv4 permission model treats ACLs separately from other extended attributes.

In FreeBSD’s implementation, a file’s owner always has read_acl, write_acl, read_attributes, and write_attributes permissions, even if the file’s ACL itself specifies otherwise.

**NFSv4 entities for which permissions can be specified**

In addition to the garden-variety user:username and group:groupname specifiers, NFSv4 defines several special entities that may be assigned permissions in an ACL. Most important among these are owner@, group@, and everyone@, which correspond to the traditional categories in the 9-bit permission model.

NFSv4 has several differences from POSIX. For one thing, it has no default entity, used in POSIX to control ACL inheritance. Instead, any individual access control entry (ACE) can be flagged as inheritable (see ACL inheritance in NFSv4, below). NFSv4 also does not use a mask to reconcile the permissions specified in a file’s mode with its ACL. The mode is required to be consistent with the settings specified for owner@, group@, and everyone@, and filesystems that implement NFSv4 ACLs must preserve this consistency when either the mode or the ACL is updated.

**NFSv4 access determination**

<table>
<thead>
<tr>
<th>Code</th>
<th>Verbose name</th>
<th>Permission</th>
</tr>
</thead>
<tbody>
<tr>
<td>r</td>
<td>read_data</td>
<td>Read data (file) or list directory contents (directory)</td>
</tr>
<tr>
<td>w</td>
<td>write_data</td>
<td>Write data (file) or create file (directory)</td>
</tr>
<tr>
<td>x</td>
<td>execute</td>
<td>Execute as a program</td>
</tr>
<tr>
<td>p</td>
<td>append_data</td>
<td>Append data (file) or create subdirectory (directory)</td>
</tr>
<tr>
<td>D</td>
<td>delete_child</td>
<td>Delete child within a directory</td>
</tr>
<tr>
<td>d</td>
<td>delete</td>
<td>Delete</td>
</tr>
<tr>
<td>a</td>
<td>read_attributes</td>
<td>Read nonextended attributes</td>
</tr>
<tr>
<td>A</td>
<td>write_attributes</td>
<td>Write nonextended attributes</td>
</tr>
<tr>
<td>R</td>
<td>read_xattr</td>
<td>Read named (“extended”) attributes</td>
</tr>
<tr>
<td>W</td>
<td>write_xattr</td>
<td>Write named (“extended”) attributes</td>
</tr>
<tr>
<td>c</td>
<td>read_acl</td>
<td>Read access control list</td>
</tr>
<tr>
<td>C</td>
<td>write_acl</td>
<td>Write access control list</td>
</tr>
<tr>
<td>o</td>
<td>write_owner</td>
<td>Change ownership</td>
</tr>
<tr>
<td>s</td>
<td>synchronize</td>
<td>Allow requests for synchronous I/O (usually ignored)</td>
</tr>
</tbody>
</table>
The NFSv4 system differs from POSIX in that an ACE specifies only a partial set of permissions. Each ACE is either an “allow” ACE or a “deny” ACE; it acts more like a mask than an authoritative specification of all possible permissions. Multiple ACEs can apply to any given situation.

When deciding whether to allow a particular operation, the filesystem reads the ACL in order, processing ACEs until either all requested permissions have been granted or some requested permission has been denied. Only ACEs whose entity strings are compatible with the current user’s identity are considered.

This iterative evaluation process means that owner@, group@, and everyone@ are not exact analogs of the corresponding traditional mode bits. An ACL can contain multiple copies of these elements, and their precedence is determined by their order of appearance in the ACL rather than by convention. In particular, everyone@ really does apply to everyone, not just users who aren’t addressed more specifically.

It’s possible for the filesystem to reach the end of an ACL without having obtained a definitive answer to a permission query. The NFSv4 standard considers the result to be undefined, but real-world implementations deny access, both because this is the convention used by Windows and because it’s the only option that makes sense.

ACL inheritance in NFSv4

Like POSIX ACLs, NFSv4 ACLs allow newly created objects to inherit access control entries from their enclosing directory. However, the NFSv4 system is a bit more powerful and a lot more confusing. Here are the important points:

- You can flag any ACE as inheritable. Inheritance for newly created subdirectories (dir_inherit or d) and inheritance for newly created files (file_inherit or f) are flagged separately.
- You can apply different access control entries to new files and new directories by creating separate access control entries on the parent directory and flagging them appropriately. You can also apply a single ACE to all new child entities (of whatever type) by turning on both the d and f flags.
- From the perspective of access determination, access control entries have the same effect on the parent (source) directory whether or not they are inheritable. If you want an entry to apply to children but not to the parent directory itself, turn on the ACE’s inherit_only (i) flag.
- New subdirectories normally inherit two copies of each ACE: one with the inheritance flags turned off, which applies to the subdirectory itself; and one with the inherit_only flag turned on, which sets up the new subdirectory to propagate its inherited ACEs. You can suppress the creation of this second ACE by turning on the no_propagate
(n) flag on the parent directory’s copy of the ACE. The end result is that the ACE propagates only to immediate children of the original directory.

- Don’t confuse the propagation of access control entries with true inheritance. Your setting an inheritance-related flag on an ACE simply means that the ACE will be copied to new entities. It does not create any ongoing relationship between the parent and its children. If you later change the ACE entries on the parent directory, the children are not updated.

Table 5.9 summarizes these various inheritance flags.

<table>
<thead>
<tr>
<th>Code</th>
<th>Verbose name</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>f</td>
<td>file_inherit</td>
<td>Propagate this ACE to newly created files.</td>
</tr>
<tr>
<td>d</td>
<td>dir_inherit</td>
<td>Propagate this ACE to newly created subdirectories.</td>
</tr>
<tr>
<td>i</td>
<td>inherit_only</td>
<td>Propagate, but don’t apply to the current directory.</td>
</tr>
<tr>
<td>n</td>
<td>no_propagate</td>
<td>Propagate to new subdirectories, but turn off inheritance.</td>
</tr>
</tbody>
</table>

**NFSv4 ACL viewing**

FreeBSD has extended the standard `setfacl` and `getfacl` commands used with POSIX ACLs to handle NFSv4 ACLs as well. For example, here’s the ACL for a newly created directory:

```
freebsd$ mkdir example
freebsd$ ls -ld example
drwxr-xr-x  2 garth staff  2 Aug 16 18:52 example/

$ getfacl -q example
    owner@:rwxp--aARWcCos:------:allow
    group@:r----a-R-c--s:------:allow
    everyone@:r----a-R-c--s:------:allow
```

The -v flag requests verbose permission names. (We indented the following output lines and wrapped them at slashes to clarify structure.)

```
freebsd$ getfacl -qv example
    owner@:read_data/write_data/execute/append_data/read_attributes/
        write_attributes/read_xattr/write_xattr/read_acl/write_acl/
        write_owner/synchronize::allow
    group@:read_data/execute/read_attributes/read_xattr/read_acl/
        synchronize::allow
    everyone@:read_data/execute/read_attributes/read_xattr/read_acl/
        synchronize::allow
```
This newly created directory seems to have a complex ACL, but in fact this is just the 9-bit mode translated into ACLese. It is not necessary for the filesystem to store an actual ACL, because the ACL and the mode are equivalent. (As with POSIX ACLS, such lists are termed “minimal” or “trivial.”) If the directory had an actual ACL, `ls` would show the mode bits with a + on the end (i.e., `drwxr-xr-x+`) to mark its presence.

Each clause represents one access control entry. The format is

```
entity:permissions:inheritance_flags:type
```

The `entity` can be the keywords `owner@`, `group@`, or `everyone@`, or a form such as `user:username` or `group:groupname`. Both the `permissions` and the `inheritance_flags` are slash-separated lists of options in the verbose output and `ls`-style bitmaps in short output. The `type` of an ACE is either `allow` or `deny`.

The use of a colon as a subdivider within the `entity` field makes it tricky for scripts to parse `getfacl` output no matter which output format you use. If you need to process ACLs programmatically, it’s best to do so through a modular API rather than by parsing command output.

**Interactions between ACLs and modes**

The mode and the ACL must remain consistent, so whenever you adjust one of these entities, the other automatically updates to conform to it. It’s easy for the system to determine the appropriate mode for a given ACL. However, emulating the traditional behavior of the mode with a series of access control entries is trickier, especially in the context of an existing ACL. The system must often generate multiple and seemingly inconsistent sets of entries for `owner@`, `group@`, and `everyone@` that depend on evaluation order for their aggregate effect.

As a rule, it’s best to avoid tampering with a file’s or directory’s mode once you’ve applied an ACL.

**NFSv4 ACL setup**

Because the permission system enforces consistency between a file’s mode and its ACL, all files have at least a trivial ACL. Ergo, ACL changes are always updates.

You make ACL changes with the `setfacl` command, much as you do under the POSIX ACL regime. The main difference is that order of access control entries is significant for an NFSv4 ACL, so you might need to insert new entries at a particular point within the existing ACL. You can do that with the `-a` flag:

```
setfacl -a position entries file ...
```

Here, `position` is the index of the existing access control entry (numbered starting at zero) in front of which the new entries should be inserted. For example, the command
installs an access control entry on the file `ben_keep_out` that denies all permissions to the user ben. The notation `full_set` is a shorthand notation that includes all possible permissions. (Written out, those would currently be `rwxpDdaARWcCos`; compare with Table 5.8.)

Because the new access control entry is inserted at position zero, it’s the first one consulted and takes precedence over later entries. Ben will be denied access to the file even if, for example, the `everyone@` permissions grant access to other users.

You can also use long names such as `write_data` to identify permissions. Separate multiple long names with slashes. You cannot mix single-letter codes and long names in a single command.

As with POSIX ACLs, you can use the `-m` flag to add new entries to the end of the existing ACL.

As for complex changes to existing ACLs, you can best achieve them by dumping the ACL to a text file, editing the access control entries in a text editor, and then reloading the entire ACL. For example:

```
$ setfacl -q file > /tmp/file.acl
$ vi /tmp/file.acl                        # Make any required changes
$ setfacl -b -M /tmp/file.acl file
```

`setfacl`'s `-b` option removes the existing ACL before adding the access control entries listed in `file.acl`. Its inclusion lets you delete entries simply by removing them from the text file.
6 Software Installation and Management

The installation, configuration, and management of software is a large part of most sysadmins’ jobs. Administrators respond to installation and configuration requests from users, apply updates to fix security problems, and supervise transitions to new software releases that may offer both new features and incompatibilities. Generally, administrators perform all the following tasks:

- Automating mass installations of operating systems
- Maintaining custom OS configurations
- Keeping systems and applications patched and up to date
- Tracking software licenses
- Managing add-on software packages

The process of configuring an off-the-shelf distribution or software package to conform to your needs (and to your local conventions for security, file placement, and network topology) is often referred to as “localization.” This chapter explores some techniques and software that help reduce the pain of software installation and make these tasks scale more gracefully. We also discuss the installation procedure for each of our example operating systems, including some options for automated deployment that use common (platform-specific) tools.
6.1 Operating system installation

Linux distributions and FreeBSD have straightforward procedures for basic installation. For physical hosts, installation typically involves booting from external USB storage or optical media, answering a few basic questions, optionally configuring disk partitions, and then telling the installer which software packages to install. Most systems, including all our example distributions, include a “live” option on the installation media that lets you run the operating system without actually installing it on a local disk.

Installing the base operating system from local media is fairly trivial thanks to the GUI applications that shepherd you through the process. Table 6.1 lists pointers to detailed installation instructions for each of our example systems.

Table 6.1: Installation documentation

<table>
<thead>
<tr>
<th>System</th>
<th>Documentation source</th>
</tr>
</thead>
<tbody>
<tr>
<td>Red Hat</td>
<td>redhat.com/docs/manuals/enterprise</td>
</tr>
<tr>
<td>CentOS</td>
<td>wiki.centos.org/Manuals/ReleaseNotes/CentOS7</td>
</tr>
<tr>
<td>Debian</td>
<td>debian.org/releases/stable/installmanual</td>
</tr>
<tr>
<td>Ubuntu</td>
<td>help.ubuntu.com/lts/serverguide/installation.html</td>
</tr>
<tr>
<td>FreeBSD</td>
<td>freebsd.org/doc/handbook/bsdinstall.html</td>
</tr>
</tbody>
</table>
Installing from the network

If you have to install an operating system on more than one computer, you will quickly reach the limits of interactive installation. It's time consuming, error prone, and boring to repeat the standard installation process on hundreds of systems. You can minimize human errors with a localization checklist, but even this measure does not remove all potential sources of variation.

To alleviate some of these problems, you can use network installation options to simplify deployments. Network installations are appropriate for sites with more than ten or so systems. The most common methods use DHCP and TFTP to boot the system sans physical media. They then retrieve the OS installation files from a network server with HTTP, NFS, or FTP.

You can set up completely hands-free installations through PXE, the Preboot eXecution Environment. This scheme is a standard from Intel that lets systems boot from a network interface. It works especially well in virtualized environments.

PXE acts like a miniature OS that sits in a ROM on your network card. It exposes its network capabilities through a standardized API for the system BIOS to use. This cooperation makes it possible for a single boot loader to netboot any PXE-enabled PC without having to supply special drivers for each network card.

See this page for more information about DHCP.

The external (network) portion of the PXE protocol is straightforward and is similar to the netboot procedures used on other architectures. A host broadcasts a DHCP “discover” request with the PXE flag turned on, and a DHCP server or proxy responds with a DHCP packet that includes PXE options (the name of a boot server and boot file). The client downloads its boot file over TFTP (or, optionally, multicast TFTP) and then executes it. The PXE boot procedure is depicted in Exhibit A.

Exhibit A: PXE boot and installation process
The DHCP, TFTP, and file servers can all be located on different hosts. The TFTP-provided boot file includes a menu with pointers to the available OS boot images, which can then be retrieved from the file server with HTTP, FTP, NFS, or some other network protocol.

PXE booting is most commonly used in conjunction with unattended installation tools such as Red Hat’s kickstart or Debian’s preseeding system, as discussed in the upcoming sections. You can also use PXE to boot diskless systems such as thin clients.

Cobbler, discussed here, includes some glue that makes netbooting much easier. However, you will still need a working knowledge of the tools that underlie Cobbler, beginning with PXE.
Setting up PXE

The most widely used PXE boot system is H. Peter Anvin’s PXELINUX, which is part of his SYSLINUX suite of boot loaders for every occasion. Check it out at syslinux.org. Another option is iPXE (ipxe.org), which supports additional bootstrapping modes, including support for wireless networks.

PXELINUX supplies a boot file that you install in the TFTP server’s tftpboot directory. To boot from the network, a PC downloads the PXE boot loader and its configuration from the TFTP server. The configuration file lists one or more options for operating systems to boot. The system can boot through to a specific OS installation without any user intervention, or it can display a custom boot menu.

PXELINUX uses the PXE API for its downloads and is therefore hardware independent all the way through the boot process. Despite the name, PXELINUX is not limited to booting Linux. You can deploy PXELINUX to install FreeBSD and other operating systems, including Windows.

See this page for more information about DHCP server software.

On the DHCP side, ISC’s (the Internet Systems Consortium’s) DHCP server is your best bet for serving PXE information. Alternatively, try Dnsmasq (goo.gl/FNk7a), a lightweight server with DNS, DHCP, and netboot support. Or simply use Cobbler, discussed below.
Using kickstart, the automated installer for Red Hat and CentOS

Kickstart is a Red Hat-developed tool for performing automated installations. It is really just a scripting interface to the standard Red Hat installer software, Anaconda, and depends both on the base distribution and on RPM packages. Kickstart is flexible and quite smart about autodetecting the system’s hardware, so it works well for bare-metal and virtual machines alike. Kickstart installs can be performed from optical media, the local hard drive, NFS, FTP, or HTTP.

Setting up a kickstart configuration file

Kickstart’s behavior is controlled by a single configuration file, generally called ks.cfg. The format of this file is straightforward. If you’re visually inclined, Red Hat’s handy GUI tool system-config-kickstart lets you point and click your way to ks.cfg nirvana.

A kickstart config file consists of three ordered parts. The first part is the command section, which specifies options such as language, keyboard, and time zone. This section also specifies the source of the distribution with the url option. In the following example, it’s a host called installserver.

Here’s an example of a complete command section:

```
text
lang en_US
langs support en_US
keyboard us
timezone --utc America/EST
mouse
rootpw --iscrypted $6$NaC1$X5jR1REy9DqNTCXjHp075/
reboot
bootloader --location=mbr
install
url --url http://installserver/redhat
clearpart --all --initlabel
part / --fstype ext3 --size 4096
part swap --size 1024
part /var --fstype ext3 -size 1 --grow
network --bootproto dhcp
auth --useshadow --enab lemd5
firewall --disabled
xconfig --defaultdesktop=GNOME --startxonboot --resolution 1280x1024 --depth 24
```

Kickstart uses graphical mode by default, which defeats the goal of unattended installation. The text keyword at the top of the example fixes this.
The `rootpw` option sets the new machine’s root password. The default is to specify the password in cleartext, which presents a serious security problem. Always use the `--iscrypted` flag to specify a hashed password. To encrypt a password for use with kickstart, use `openssl passwd -1`. Still, this option leaves all your systems with the same root password. Consider running a postboot process to change the password at build time.

The `clearpart` and `part` directives specify a list of disk partitions and their sizes. You can include the `-grow` option to expand one of the partitions to fill any remaining space on the disk. This feature makes it easy to accommodate systems that have different sizes of hard disk. Advanced partitioning options, such as the use of LVM, are supported by kickstart but not by the `system-config-kickstart` tool. Refer to Red Hat’s on-line documentation for a complete list of disk layout options.

The second section is a list of packages to install. It begins with a `%packages` directive. The list can contain individual packages, collections such as `@ GNOME`, or the notation `@ Everything` to include the whole shebang. When selecting individual packages, specify only the package name, not the version or the `.rpm` extension. Here’s an example:

```
%packages
@ Networked Workstation
@ X Window System
@ GNOME
mylocalpackage
```

In the third section of the kickstart configuration file, you can specify arbitrary shell commands for kickstart to execute. There are two possible sets of commands: one introduced with `%pre` that runs before installation, and one introduced with `%post` that runs afterward. Both sections have some restrictions on the ability of the system to resolve hostnames, so it’s safest to use IP addresses if you want to access the network. In addition, the postinstall commands are run in a `chroot`ed environment, so they cannot access the installation media.

The `ks.cfg` file is quite easy to generate programmatically. One option is to use the pykickstart Python library, which can read and write kickstart configurations.

For example, suppose you wanted to install different sets of packages on servers and clients and that you also have two separate physical locations that require slightly different customizations. You could use pykickstart to write a script that transforms a master set of parameters into a set of four separate configuration files, one for servers and one for clients in each office.

Changing the complement of packages would then be just a matter of changing the master configuration file rather than of changing every possible configuration file. There may even be cases in which you need to generate individualized configuration files for specific hosts. In this situation, you would certainly want the final `ks.cfg` files to be automatically generated.

**Building a kickstart server**
Kickstart expects its installation files, called the installation tree, to be laid out as they are on the distribution media, with packages stored in a directory called `RedHat/RPMS` on the server. If you’re installing over the network via FTP, NFS, or HTTP, you can either copy the contents of the distribution (leaving the tree intact), or you can simply use the distribution’s ISO images. You can also add your own packages to this directory. There are, however, a couple of issues to be aware of.

First, if you tell kickstart to install all packages (with an `@ Everything` in the packages section of your `ks.cfg`), it installs add-on packages in alphabetical order once all the base packages have been laid down. If your package depends on other packages that are not in the base set, you might want to call your package something like `zzmypackage.rpm` to make sure that it’s installed last.

If you don’t want to install all packages, either list your supplemental packages individually in the `%packages` section of the `ks.cfg` file or add your packages to one or more of the collection lists. Collection lists are specified by entries such as `@ GNOME` and stand for a predefined set of packages whose members are enumerated in the file `RedHat/base/comps` on the server. The collections are the lines that begin with 0 or 1; the number specifies whether the collection is selected by default.

In general, it’s not a good idea to tamper with the standard collections. Leave them as Red Hat defined them and explicitly name all your supplemental packages in the `ks.cfg` file.

**Pointing kickstart at your config file**

See [this page](#) for more information about PXE.

Once you’ve created a config file, you have a couple of ways to get kickstart to use it. The officially sanctioned method is to boot from external media (USB or DVD) and ask for a kickstart installation by specifying `linux inst.ks` at the initial `boot:` prompt. PXE boot is also an option.

If you don’t specify additional arguments, the system determines its network address with DHCP. It then obtains the DHCP boot server and boot file options, attempts to mount the boot server with NFS, and uses the value of the boot file option as its kickstart configuration file. If no boot file has been specified, the system looks for a file called `/kickstart/host_ip_address-kickstart`.

Alternatively, you can tell kickstart to get its configuration file in some other way by supplying a path as an argument to the `inst.ks` option. There are several possibilities. For example, the instruction

```
boot: linux inst.ks=http://server:/path
```
tells kickstart to use HTTP to download the file instead of NFS. (Prior to RHEL 7, the option was called **ks** instead of **inst.ks**. Both are understood for now, but future versions may drop **ks**.)

To eliminate the use of boot media entirely, you’ll need to graduate to PXE. See [this page](#) for more information about that.
Automating installation for Debian and Ubuntu

Debian and Ubuntu can use the **debian-installer** for “preseeding,” the recommended method for automated installation. As with Red Hat’s kickstart, a preconfiguration file answers questions asked by the installer.

All the interactive parts of the Debian installer use the **debconf** utility to decide which questions to ask and what default answers to use. By giving **debconf** a database of preformulated answers, you fully automate the installer. You can either generate the database by hand (it’s a text file), or you can perform an interactive installation on an example system and then dump out your **debconf** answers with the following commands:

```bash
$ sudo debconf-get-selections --installer > preseed.cfg
$ sudo debconf-get-selections >> preseed.cfg
```

Make the config file available on the net and then pass it to the kernel at installation time with the following kernel argument:

```
preserve/url=http://host/path/to/preseed
```

The syntax of the preseed file, usually called **preseed.cfg**, is simple and is reminiscent of Red Hat’s **ks.cfg**. The sample below has been shortened for simplicity.

```plaintext
d-i debian-installer/locale string en_US
d-i console-setup/ask_detect boolean false
d-i console-setup/layoutcode string us
d-i netcfg/choose_interface select auto
d-i netcfg/get_hostname string unassigned-hostname
d-i netcfg/get_domain string unassigned-domain
...
d-i partman-auto/disk string /dev/sda
d-i partman-auto/method string lvm
d-i partman-auto/choose_recipe select atomic
...
d-i passwd/user-fullname string Daffy Duck
d-i passwd/username string dduck
d-i passwd/user-password-crypted password $6$mkq9$G//i6tN.x6670.951VSM/
d-i user-setup/encrypt-home boolean false
tasksel tasksel/first multiselect ubuntu-desktop
d-i grub-installer/only_debian boolean true
d-i grub-installer/with_other_os boolean true
d-i finish-install/reboot_in_progress note
xserver-xorg xserver-xorg/autodetect_monitor boolean true
...
```
Several options in this list simply disable dialogs that would normally require user interaction. For example, the `console-setup/ask_detect` clause disables manual keymap selection.

This configuration tries to identify a network interface that’s actually connected to a network (`choose_interface select auto`) and obtains network information through DHCP. The system hostname and domain values are presumed to be furnished by DHCP and are not overridden.

Preseeded installations cannot use existing partitions; they must either use existing free space or repartition the entire disk. The `partman*` lines in the code above are evidence that the `partman-auto` package is being used for disk partitioning. You must specify a disk to install to unless the system has only one. In this case, `/dev/sda` is used.

Several partitioning recipes are available.

- **atomic** puts all the system’s files in one partition.
- **home** creates a separate partition for `/home`.
- **multi** creates separate partitions for `/home`, `/usr`, `/var`, and `/tmp`.

You can create users with the `passwd` series of directives. As with kickstart configuration, we strongly recommend the use of encrypted (hashed) password values. Preseed files are often stored on HTTP servers and are apt to be discovered by curious users. (Of course, a hashed password is still subject to brute force attack. Use a long, complex password.)

The task selection (`tasksel`) option chooses the type of Ubuntu system to install. Available values include `standard`, `ubuntu-desktop`, `dns-server`, `lamp-server`, `kubuntu-desktop`, `edubuntu-desktop`, and `xubuntu-desktop`.

The sample preseed file shown above comes from the Ubuntu installation documentation found at help.ubuntu.com. This guide contains full documentation for the syntax and usage of the preseed file.

Although Ubuntu does not descend from the Red Hat lineage, it has grafted compatibility with kickstart control files onto its own underlying installer. Ubuntu also includes the `system-config-kickstart` tool for creating these files. However, the kickstart functionality in Ubuntu’s installer is missing a number of important features that are supported by Red Hat’s Anaconda, such as LVM and firewall configuration. We recommend sticking with the Debian installer unless you have a good reason to choose kickstart (e.g., to maintain compatibility with your Red Hat systems).
Netbooting with Cobbler, the open source Linux provisioning server

By far the easiest way to bring netbooting services to your network is with Cobbler, a project originally written by Michael DeHaan, prolific open source developer. Cobbler enhances kickstart to remove some of its most tedious and repetitive administrative elements. It bundles all the important netboot features, including DHCP, DNS, and TFTP, and helps you manage the OS images used to build physical and virtual machines. Cobbler includes command-line and web interfaces for administration.

Templates are perhaps Cobbler’s most interesting and useful feature. You’ll frequently need different kickstart and preseed settings for different host profiles. For example, you might have web servers in two data centers that, apart from network settings, require the same configuration. You can use Cobbler “snippets” to share sections of the configuration between the two types of hosts.

A snippet is just a collection of shell commands. For example, this snippet adds a public key to the authorized SSH keys for the root user:

```bash
mkdir -p --mode=700 /root/.ssh
cat >> /root/.ssh/authorized_keys << EOF
ssh-rsa AAAAB3NzaC1yc2EAAAADAQABAAABAQDKErzVdanNkL4bzAZotSzU/
 ... Rooy2R6TCzc1Bt/oqUK1R1kuV
EOF
chmod 600 /root/.ssh/authorized_keys
```

You save the snippet to Cobbler’s snippet directory, then refer to it in a kickstart template. For example, if you saved the snippet above as `root_pubkey_snippet`, you could refer to it in a template as follows.

```bash
%post
SNIPPET::root_pubkey_snippet
$kickstart_done
```

Use Cobbler templates to customize disk partitions, conditionally install packages, customize time zones, add custom package repositories, and perform any other kind of localization requirement.

Cobbler can also create new virtual machines under a variety of hypervisors. It can integrate with a configuration management system to provision machines once they boot.

Cobbler packages are available in the standard repositories for our sample Linux distributions. You can also obtain packages and documentation from the Cobbler GitHub project at cobbler.github.io.
Automating FreeBSD installation

The FreeBSD `bsdinstall` utility is a text-based installer that kicks off when you boot a computer from a FreeBSD installation CD or DVD. Its automation facilities are rudimentary compared to Red Hat’s kickstart or Debian’s preseed, and the documentation is limited. The best source of information is the `bsdinstall` man page.

Creating a customized, unattended installation image is a tedious affair that involves the following steps.

1. Download the latest installation ISO (CD image) from ftp.freebsd.org.
2. Unpack the ISO image to a local directory.
3. Make any desired edits in the cloned directory.
4. Create a new ISO image from your customized layout and burn it to media, or create a PXE boot image for netbooting.

FreeBSD’s version of `tar` understands ISO format in addition to many other formats, so you can simply extract the CD image files to a scratch directory. Create a subdirectory before extracting, because the ISO file unpacks to the current directory.

```
freebsd$ sudo mkdir FreeBSD
freebsd$ sudo tar xpCf FreeBSD FreeBSD-11.0.iso
```

Once you’ve extracted the contents of the image, you can customize them to reflect your desired installation settings. For example, you could add custom DNS resolvers by editing `FreeBSD/etc/resolv.conf` to include your own name servers.

`bsdinstall` normally requires users to select settings such as the type of terminal in use, the keyboard mapping, and the desired style of disk partitioning. You can bypass the interactive questions by putting a file called `installerconfig` in the `etc` directory of the system image.

This file’s format is described in the `bsdinstall` man page. It has two sections:

- The preamble, which sets certain installation settings
- A shell script which executes after installation completes

We refer you to the man page rather than regurgitate its contents here. Among other settings, it contains options for installing directly to a ZFS root and to other custom partitioning schemes.

Once your customizations are complete, you can create a new ISO file with the `mkisofs` command. Create a PXE image or burn the ISO to optical media for an unattended installation.
The mfsBSD project (mfsbsd.vx.sk) is a set of scripts that generate a PXE-friendly ISO image. The basic FreeBSD 11 image weighs in at a lean 47MiB. See the source scripts at github.com/mmatuska/mfsbsd.
6.2 Managing Packages

UNIX and Linux software assets (source code, build files, documentation, and configuration templates) were traditionally distributed as compressed archives, usually gzipped tarballs (.tar.gz or .tgz files). This was OK for developers but inconvenient for end users and administrators. These source archives had to be manually compiled and built for each system on each release of the software, a tedious and error prone process.

Packaging systems emerged to simplify and facilitate the job of software management. Packages include all the files needed to run a piece of software, including precompiled binaries, dependency information, and configuration file templates that can be customized by administrators. Perhaps most importantly, packaging systems try to make the installation process as atomic as possible. If an error occurs during installation, a package can be backed out or reapplied. New versions of software can be installed with a simple package update.

Package installers are typically aware of configuration files and will not normally overwrite local customizations made by a system administrator. They either back up existing config files that they change or supply example config files under a different name. If you find that a newly installed package breaks something on your system, you can, theoretically, back it out to restore your system to its original state. Of course, theory != practice, so don’t try this on a production system without testing it first.

Packaging systems define a dependency model that allows package maintainers to ensure that the libraries and support infrastructure on which their applications depend are properly installed. Unfortunately, the dependency graphs are sometimes imperfect. Unlucky administrators can find themselves in package dependency hell, a state where it’s impossible to update a package because of version incompatibilities among its dependencies. Fortunately, recent versions of packaging software seem to be less susceptible to this effect.

Packages can run scripts at various points during the installation, so they can do much more than just disgorge new files. Packages frequently add new users and groups, run sanity checks, and customize settings according to the environment.

Confusingly, package versions do not always correspond directly to the versions of the software that they install. For example, consider the following RPM package for docker-engine:

```
$ rpm -qa | grep -i docker
docker-engine-1.13.0-1.el7.centos.x86_64
$ docker version | grep Version
Version: 1.13.1
```

The package itself claims version 1.13.0, but the docker binary reports version 1.13.1. In this case, the distribution maintainers backported changes and incremented the minor package
version. Be aware that the package version string is not necessarily an accurate indication of the software version that is actually installed.

You can create packages to facilitate the distribution of your own localizations or software. For example, you can create a package that, when installed, reads localization information for a machine (or gets it from a central database) and uses that information to set up local configuration files.

You can also bundle local applications as packages (complete with dependencies) or create packages for third party applications that aren’t normally distributed in package format. You can version your packages and use the dependency mechanism to upgrade machines automatically when a new version of your localization package is released. We refer you to **fpm**, the Effing Package Manager, which is the easiest way to get started building packages for multiple platforms. You can find it at [github.com/jordansissel/fpm](http://github.com/jordansissel/fpm).

You can also use the dependency mechanism to create groups of packages. For example, you can create a package that installs nothing of its own but depends on many other packages. Installing the package with dependencies turned on results in all the packages being installed in a single step.
6.3 Linux package management systems

Two package formats are in common use on Linux systems. Red Hat, CentOS, SUSE, Amazon Linux, and several other distributions use RPM, a recursive acronym that expands to “RPM Package Manager.” Debian and Ubuntu use the separate but equally popular .deb format. The two formats are functionally similar.

Both the RPM and .deb packaging systems now function as dual-layer soup-to-nuts configuration management tools. At the lowest level are the tools that install, uninstall, and query packages: `rpm` for RPM and `dpkg` for .deb.

On top of these commands are systems that know how to find and download packages from the Internet, analyze interpackage dependencies, and upgrade all the packages on a system. `yum`, the Yellowdog Updater, Modified, works with the RPM system. APT, the Advanced Package Tool, originated in the .deb universe but works well with both .deb and RPM packages.

On the next couple of pages, we review the low-level commands `rpm` and `dpkg`. In the section High-level Linux package management systems, we discuss the comprehensive update systems APT and yum, which build on these low-level facilities. Your day-to-day administration activities will usually involve the high-level tools, but you’ll occasionally need to wade into the deep end of the pool with `rpm` and `dpkg`. 
rpm: manage RPM packages

The `rpm` command installs, verifies, and queries the status of packages. It formerly built them as well, but this function has now been relegated to a separate command called `rpmbuild`. `rpm` options have complex interactions and can be used together only in certain combinations. It’s most useful to think of `rpm` as if it were several different commands that happen to share the same name.

The mode you tell `rpm` to enter (such as `-i` or `-q`) specifies which of `rpm`’s multiple personalities you are hoping to access. `rpm --help` lists all the options broken down by mode, but it’s worth your time to read the man page in some detail if you will frequently be dealing with RPM packages.

The bread-and-butter options are `-i` (install), `-U` (upgrade), `-e` (erase), and `-q` (query). The `-q` option is a bit tricky; you must supply an additional command-line flag to pose a specific question. For example, the command `rpm -qa` lists all the packages installed on the system.

Let’s look at an example. We need to install a new version of OpenSSH because of a recent security fix. Once we’ve downloaded the package, we’ll run `rpm -U` to replace the older version with the newer.

```
redhat$ sudo rpm -U openssh-6.6.1p1-33.e17_2.x86_64.rpm
error: failed dependencies:
openssh = 6.6.1p1-23 is needed by openssh-clients-6.6.1p1-23
openssh = 6.6.1p1-23 is needed by openssh-server-6.6.1p1-23
```

D’oh! Perhaps it’s not so simple after all. Here we see that the currently installed version of OpenSSH, 6.6.1p1-23, is required by several other packages. `rpm` won’t let us upgrade OpenSSH to 6.6.1p1-33 because the change might affect the operation of these other packages. This type of conflict happens all the time, and it’s a major motivation for the development of systems like APT and `yum`. In real life we wouldn’t attempt to untangle the dependencies by hand, but let’s continue with `rpm` alone for the purpose of this example.

We could force the upgrade with the `--force` option, but that’s usually a bad idea. The dependency information is there to save time and trouble, not just to get in the way. There’s nothing like a broken SSH on a remote system to ruin a sysadmin’s morning.

Instead, we’ll grab updated versions of the dependent packages as well. If we were smart, we could have determined that other packages depended on OpenSSH before we even attempted the upgrade:

```
redhat$ rpm -q --whatrequires openssh
openssh-server-6.6.1p1-23.e17_2.x86_64
openssh-clients-6.6.1p1-23.e17_2.x86_64
```
Suppose that we’ve obtained updated copies of all the packages. We could install them one at a time, but `rpm` is smart enough to handle them all at once. If multiple RPMs are listed on the command line, `rpm` sorts them by dependency before installation.

```
redhat$ sudo rpm -U openssh-*
...
redhat$ rpm -q openssh
openssh-6.6.1p1-33.el7_3
```

Cool! Looks like it succeeded. Note that `rpm` understands which package we are talking about even though we didn’t specify the package’s full name or version. (Unfortunately, `rpm` does not restart `sshd` after the installation. You’d need to manually restart it to complete the upgrade.)
dpkg: manage .deb packages

Just as RPM packages have the all-in-one `rpm` command, Debian packages have the `dpkg` command. Useful options include `--install`, `--remove`, and `-l` to list the packages that have been installed on the system. A `dpkg --install` of a package that’s already on the system removes the previous version before installing.

Running `dpkg -l | grep package` is a convenient way to determine if a particular package is installed. For example, to search for an HTTP server, try

```
ubuntu$ dpkg -l | grep http
```

```
ii lighttpd 1.4.35-4+deb8u1 amd64 fast webserver with minimal memory footprint
```

This search found the `lighttpd` software, an excellent, open source, lightweight web server. The leading `ii` indicates that the software is installed.

Suppose that the Ubuntu security team recently released a fix to `nvi` to patch a potential security problem. After grabbing the patch, we run `dpkg` to install it. As you can see, it’s much chattier than `rpm` and tells us exactly what it’s doing.

```
ubuntu$ sudo dpkg --install ./nvi_1.81.6-12_amd64.deb
(Reading database ... 24368 files and directories currently installed.)
Preparing to replace nvi 1.79-14 (using ./nvi_1.81.6-12_amd64.deb) ...
Unpacking replacement nvi ...
Setting up nvi (1.81.6-12) ...
Checking available versions of ex, updating links in /etc/alternatives ...
(You may modify the symlinks there yourself if desired – see 'man ln'.)
Leaving ex (/usr/bin/ex) pointing to /usr/bin/nex.
Leaving ex.1.gz (/usr/share/man/man1/ex.1.gz) pointing to /usr/share/man/man1/nex.1.gz.
...
```

We can now use `dpkg -l` to verify that the installation worked. The `-l` flag accepts an optional prefix pattern to match, so we can just search for `nvi`.

```
ubuntu$ dpkg -l nvi
 Name   Version Description
ii  nvi   1.81.6-12  4.4BSD re-implementation of vi.
```

Our installation seems to have gone smoothly.
6.4 **High-level Linux package management systems**

Metapackage management systems such as APT and **yum** share several goals:

- To simplify the task of locating and downloading packages
- To automate the process of updating or upgrading systems
- To facilitate the management of interpackage dependencies

Clearly, these systems include more than just client-side commands. They all require that distribution maintainers organize their offerings in an agreed-on way so that the software can be accessed and reasoned about by clients.

Since no single supplier can encompass the entire “world of Linux software,” the systems all allow for the existence of multiple software repositories. Repositories can be local to your network, so these systems make a dandy foundation for creating your own internal software distribution system.

**RHEL**  The Red Hat Network is closely tied to Red Hat Enterprise Linux. It’s a commercial service that costs money and offers more in terms of attractive GUIs, site-wide system management, and automation ability than do APT and **yum**. It is a shiny, hosted version of Red Hat’s expensive and proprietary Satellite Server. The client side can reference **yum** and APT repositories, and this ability has allowed distributions such as CentOS to adapt the client GUI for nonproprietary use.

APT is better documented than the Red Hat Network, is significantly more portable, and is free. It’s also more flexible in terms of what you can do with it. APT originated in the world of Debian and **dpkg**, but it has been extended to encompass RPMs, and versions that work with all our example distributions are available. It’s the closest thing we have at this point to a universal standard for software distribution.

**yum** is an RPM-specific analog of APT. It’s included by default on Red Hat Enterprise Linux and CentOS, although it runs on any RPM-based system, provided that you can point it toward appropriately formatted repositories.

We like APT and consider it a solid choice if you run Debian or Ubuntu and want to set up your own automated package distribution network. See the section [APT: the Advanced Package Tool](#) for more information.
Package repositories

Linux distributors maintain software repositories that work hand-in-hand with their chosen package management systems. The default configuration for the package management system usually points to one or more well-known web or FTP servers that are under the distributor’s control.

However, it isn’t immediately obvious what such repositories should contain. Should they include only the sets of packages blessed as formal, major releases? Formal releases plus current security updates? Up-to-date versions of all the packages that existed in the formal releases? Useful third party software not officially supported by the distributor? Source code? Binaries for multiple hardware architectures? When you run `apt upgrade` or `yum upgrade` to bring the system up to date, what exactly should that mean?

In general, package management systems must answer all these questions and must make it easy for sites to select the cross-sections they want to include in their software “world.” The following concepts help structure this process.

- A “release” is a self-consistent snapshot of the package universe. Before the Internet era, named OS releases were more or less immutable and were associated with one specific time; security patches were made available separately. These days, a release is a more nebulous concept. Releases evolve over time as packages are updated. Some releases, such as Red Hat Enterprise Linux, are specifically designed to evolve slowly; by default, only security updates are incorporated. Other releases, such as beta versions, change frequently and dramatically. But in all cases, the release is the baseline, the target, the “thing I want to update my system to look like.”

- A “component” is a subset of the software within a release. Distributions partition themselves differently, but one common distinction is that between core software blessed by the distributor and extra software made available by the broader community. Another distinction that’s common in the Linux world is the one between the free, open source portions of a release and the parts that are tainted by some kind of restrictive licensing agreement.

  Of particular note from an administrative standpoint are minimally active components that include only security fixes. Some releases allow you to combine a security component with an immutable baseline component to create a relatively stable version of the distribution, even though the mainline distribution may evolve much faster.

- An “architecture” represents a class of hardware. The expectation is that machines within an architecture class are similar enough that they can all run the same binaries. Architectures are instances of releases, for example, “Ubuntu Xenial Xerus for
x86_64.” Since components are subdivisions of releases, there’s a corresponding architecture-specific instance for each of them as well.

- Individual packages are the elements that make up components, and therefore, indirectly, releases. Packages are usually architecture-specific and are versioned independently of the main release and of other packages. The correspondence between packages and releases is implicit in the way the network repository is set up.

The existence of components that aren’t maintained by the distributor (e.g., Ubuntu’s “universe” and “multiverse”) raises the question of how these components relate to the core OS release. Can they really be said to be “a component” of the specific release, or are they some other kind of animal entirely?

From a package management perspective, the answer is clear: extras are a true component. They are associated with a specific release, and they evolve in tandem with it. The separation of control is interesting from an administrative standpoint, but it doesn’t affect the package distribution systems, except that multiple repositories might need to be manually added by the administrator.
**RHN: the Red Hat Network**

With Red Hat having gracefully departed from the consumer Linux business, the Red Hat Network has become the system management platform for Red Hat Enterprise Linux. You purchase the right to access the Red Hat Network by subscribing. At its simplest, you can use the Red Hat Network as a glorified web portal and mailing list. Used in this way, the Red Hat Network is not much different from the patch notification mailing lists that have been run by various UNIX vendors for years. But more features are available if you’re willing to pay for them. For current pricing and features, see rhn.redhat.com.

The Red Hat Network presents a web-based interface for downloading new packages as well as a command-line alternative. Once you register, your machines get all the patches and bug fixes that they need without your ever having to intervene.

The downside of automatic registration is that Red Hat decides what updates you need. You might consider how much you really trust Red Hat (and the software maintainers whose products they package) not to screw things up.

A reasonable compromise might be to sign up one machine in your organization for automatic updates. You can take snapshots from that machine at periodic intervals to test as possible candidates for internal releases.
APT: the Advanced Package Tool

APT is one of the most mature package management systems. It’s possible to upgrade an entire system full of software with a single `apt` command or even (as with the Red Hat Network) to have your boxes continuously keep themselves up to date without human intervention.

The first rule of using APT on Ubuntu systems (and indeed all management of Debian packages) is to ignore the existence of `dselect`, which acts as a front end for the Debian package system. It’s not a bad idea, but the user interface is poor and can be intimidating to the novice user. Some documentation will try to steer you toward `dselect`, but stay strong and stick with `apt`.

If you are using APT to manage a stock Ubuntu installation from a standard repository mirror, the easiest way to see the available packages is to visit the master list at `packages.ubuntu.com`. The web site includes a nice search interface. If you set up your own APT server (see this page), then of course you will know what packages you have made available and you can list them in whatever way you want.

Distributions commonly include dummy packages that exist only to claim other packages as prerequisites. `apt` downloads and upgrades prerequisite packages as needed, so the dummy packages make it easy to install or upgrade several packages as a block. For example, installing the `gnome-desktop-environment` package obtains and installs all the packages necessary to run the GNOME UI.

APT includes a suite of low-level commands like `apt-get` and `apt-cache` that are wrapped for most purposes by an omnibus `apt` command. The wrapper is a later addition to the system, so you’ll still see occasional references to the low-level commands on the web and in documentation. To a first approximation, commands that look similar are in fact the same command. There’s no difference between `apt install` and `apt-get install`, for example.

Once you have set up your `/etc/apt/sources.list` file (described in detail below) and know the name of a package that you want, the only remaining task is to run `apt update` to refresh `apt`’s cache of package information. After that, just run `apt install package-name` as a privileged user to install the package. The same command updates a package that has already been installed.

Suppose you want to install a new version of the `sudo` package that fixes a security bug. First, it’s always wise to do an `apt update`:

```bash
debian$ sudo apt update
Get:1 http://http.us.debian.org stable/main Packages [824kB]
...
```

Now you can actually fetch the package. Note the use of `sudo` to fetch the new `sudo` package — `apt` can even upgrade packages that are in use!
debian$ sudo apt install sudo
Reading Package Lists... Done
Building Dependency Tree... Done
1 packages upgraded, 0 newly installed, 0 to remove and 191 not upgraded.
Need to get 0B/122kB of archives. After unpacking 131kB will be used.
(Reading database ... 24359 files and directories currently installed.)
Preparing to replace sudo 1.6.2p2-2 (using .../sudo_1.8.10p3-1+deb8u3_amd64.deb) ...
Unpacking replacement sudo ...
Setting up sudo (1.8.10p3-1+deb8u3) ...
Installing new version of config file /etc/pam.d/sudo ...
Repository configuration

Configuring APT is straightforward; pretty much everything you need to know can be found in Ubuntu’s community documentation on package management:

help.ubuntu.com/community/AptGet/Howto

The most important configuration file is `/etc/apt/sources.list`, which tells APT where to get its packages. Each line specifies the following:

- A type of package, currently `deb` or `deb-src` for Debian-style packages or `rpm` or `rpm-src` for RPMs
- A URL that points to a file, HTTP server, or FTP server from which to fetch packages
- A “distribution” (really, a release name) that lets you deliver multiple versions of packages
- A potential list of components (categories of packages within a release)

Unless you want to set up your own APT repository or cache, the default configuration generally works fine. Source packages are downloaded from the entries beginning with `deb-src`.

On Ubuntu systems, you’ll almost certainly want to include the “universe” component, which accesses the larger world of Linux open source software. The “multiverse” packages include non-open-source content, such as some VMware tools and components.

As long as you’re editing the `sources.list` file, you may want to retarget the individual entries to point to your closest mirror. A full list of Ubuntu mirrors can be found at launchpad.net/ubuntu/+archivemirrors. This is a dynamic (and long) list of mirrors that changes regularly, so be sure to keep an eye on it between releases.

Make sure that security.ubuntu.com is listed as a source so that you have access to the latest security patches.
An example /etc/apt/sources.list file

The following example uses archive.ubuntu.com as a package source for the “main” components of Ubuntu (those that are fully supported by the Ubuntu team). In addition, this sources.list file includes unsupported but open source “universe” packages, and non-free, unsupported packages in the “multiverse” component. There is also a repository for updates or bug-fixed packages in each component. Finally, the last six lines are for security updates.

```
# General format: type uri distribution [components]
deb http://archive.ubuntu.com/ubuntu xenial main restricted
deb-src http://archive.ubuntu.com/ubuntu xenial main restricted
deb http://archive.ubuntu.com/ubuntu xenial-updates main restricted
deb-src http://archive.ubuntu.com/ubuntu xenial-updates main restricted
deb http://archive.ubuntu.com/ubuntu xenial universe
deb-src http://archive.ubuntu.com/ubuntu xenial universe
deb http://archive.ubuntu.com/ubuntu xenial-updates universe
deb-src http://archive.ubuntu.com/ubuntu xenial-updates universe
deb http://archive.ubuntu.com/ubuntu xenial multiverse
deb-src http://archive.ubuntu.com/ubuntu xenial multiverse
deb http://archive.ubuntu.com/ubuntu xenial-updates multiverse
deb-src http://archive.ubuntu.com/ubuntu xenial-updates multiverse
deb http://archive.ubuntu.com/ubuntu xenial-backports main restricted
    universe multiverse
deb-src http://archive.ubuntu.com/ubuntu xenial-backports main restricted
    universe multiverse
deb http://security.ubuntu.com/ubuntu xenial-security main restricted
deb-src http://security.ubuntu.com/ubuntu xenial-security main restricted
deb http://security.ubuntu.com/ubuntu xenial-security universe
deb-src http://security.ubuntu.com/ubuntu xenial-security universe
deb http://security.ubuntu.com/ubuntu xenial-security multiverse
deb-src http://security.ubuntu.com/ubuntu xenial-security multiverse
```

The distribution and components fields help APT navigate the filesystem hierarchy of the Ubuntu repository, which has a standardized layout. The root distribution is the working title given to each release, such as trusty, xenial, or yakkety. The available components are typically called main, universe, multiverse, and restricted. Add the universe and multiverse repositories only if you are comfortable having unsupported (and license-restricted, in the case of multiverse) software in your environment.

After you update the sources.list file, run apt-get update to force APT to react to your changes.
Creation of a local repository mirror

If you plan to use **apt** on a large number of machines, you will probably want to cache packages locally. Downloading a copy of each package for every machine is not a sensible use of external bandwidth. A mirror of the repository is easy to configure and convenient for local administration. Just make sure to keep it updated with the latest security patches.

The best tool for the job is the handy **apt-mirror** package, which is available from apt-mirror.github.io. You can also install the package from the **universe** component with **sudo apt install apt-mirror**.

Once installed, **apt-mirror** drops a file called **mirror.list** in `/etc/apt`. It’s a shadow version of **sources.list**, but it’s used only as a source for mirroring operations. By default, **mirror.list** conveniently contains all the repositories for the running version of Ubuntu.

To actually mirror the repositories in **mirror.list**, just run **apt-mirror** as root:

```bash
ubuntu$ sudo apt-mirror
Downloading 162 index files using 20 threads...
Begin time: Sun Feb 5 22:34:58 2017
[20]... [19]... [18]... [17]... [16]... [15]... [14]...
```

By default, **apt-mirror** puts its repository copies in `/var/spool/apt-mirror`. Feel free to change this by uncommenting the `set base_path` directive in **mirror.list**, but be aware that you must then create **mirror**, **skel**, and **var** subdirectories under the new mirror root.

**apt-mirror** takes a long time to run on its first pass because it is mirroring many gigabytes of data (currently ~40GB per Ubuntu release). Subsequent executions are faster and should be run automatically out of **cron**. You can run the **clean.sh** script from the **var** subdirectory of your mirror to clean out obsolete files.

To start using your mirror, share the base directory through HTTP, using a web server of your choice. We like to use symbolic links to the web root. For instance:

```bash
ln -s /var/spool/apt-mirror/us.archive.ubuntu.com/ubuntu /var/www/ubuntu
```

To make clients use your local mirror, edit their **sources.list** files just as if you were selecting a nonlocal mirror.
APT automation

Use cron to schedule regular apt runs. Even if you don’t install packages automatically, you may want to run apt update regularly to keep your package summaries up to date.

apt upgrade downloads and installs new versions of any packages that are currently installed on the local machine. Note that apt upgrade is defined slightly differently from the low-level command apt-get upgrade, but apt upgrade is usually what you want. (It’s equivalent to apt-get dist-upgrade --with-new-pkgs.) apt upgrade might want to delete some packages that it views as irreconcilably incompatible with the upgraded system, so be prepared for potential surprises.

If you really want to play with fire, have machines perform the upgrade in an unattended fashion by including the -y option to apt upgrade. It answers any confirmation questions that apt might ask with an enthusiastic “Yes!” Be aware that some updates, such as kernel packages, might not take effect until after a system reboot.

It’s probably not a good idea to perform automated upgrades directly from a distribution’s mirror. However, in concert with your own APT servers, packages, and release control system, this is a perfect way to keep clients in sync. A one-liner like the following keeps a box up to date with its APT server.

# apt update && apt upgrade -y

Use this command in a cron job if you want it to run on a regular schedule. You can also refer to it from a system startup script to make the machine update at boot time. See this page for more information about cron; see Chapter 2, Booting and System Management Daemons, for more information about startup scripts.

If you run updates out of cron on many machines, it’s a good idea to use time randomization to make sure that everyone doesn’t try to update at once.

If you don’t quite trust your source of packages, consider automatically downloading all changed packages without installing them. Use apt’s --download-only option to request this behavior, then review the packages by hand and install the ones you want to update. Downloaded packages are put in /var/cache/apt, and over time this directory can grow to be quite large. Clean out the unused files from this directory with apt-get autoclean.
yum: release management for RPM

yum, the Yellowdog Updater, Modified, is a metapackage manager based on RPM. It may be a bit unfair to call yum an APT clone, but it’s thematically and implementationally similar, although cleaner and slower in practice.

On the server-side, the yum-arch command compiles a database of header information from a large set of packages (often an entire release). The header database is then shared along with the packages through HTTP. Clients use the yum command to fetch and install packages; yum figures out dependency constraints and does whatever additional work is needed to complete the installation of the requested packages. If a requested package depends on other packages, yum downloads and installs those packages as well.

The similarities between apt and yum extend to the command-line options they understand. For example, yum install foo downloads and installs the most recent version of the foo package (and its dependencies, if necessary). There is at least one treacherous difference, though: apt update refreshes apt’s package information cache, but yum update updates every package on the system (it’s analogous to apt upgrade). To add to the confusion, yum upgrade is the same as yum update but with obsolescence processing enabled.

yum does not match on partial package names unless you include globbing characters (such as * and ?) to explicitly request this behavior. For example, yum update 'lib*' refreshes all packages whose name starts with “lib”. Remember to quote the globbing characters so the shell doesn’t interfere with them.

Unlike apt, yum defaults to validating its package information cache against the contents of the network repository every time you run it. Use the -C option to prevent the validation and yum makecache to update the local cache (it takes awhile to run). Unfortunately, the -C option doesn’t do much to improve yum’s sluggish performance.

yum’s configuration file is /etc/yum.conf. It includes general options and pointers to package repositories. Multiple repositories can be active at once, and each repository can be associated with multiple URLs.

A replacement for yum called DNF (for Dandified Yum) is under active development. It’s already the default package manager for Fedora and will eventually replace yum completely. DNF sports better dependency resolution and an improved API, among other features. Visit dnf.baseurl.org to learn more.
6.5 **FreeBSD software management**

FreeBSD has had packaging facilities for several releases, but it’s only now transitioning to a completely package-centric distribution model in which most elements of the core OS are defined as packages. FreeBSD’s recent releases have segregated software into three general categories:

- A “base system,” which includes a bundled set of core software and utilities
- A set of binary packages managed with the `pkg` command
- A separate “ports” system which downloads source code, applies FreeBSD-specific patches, then builds and installs it

As of FreeBSD 11, the lines between these territories have become even more muddled. The base system has been packagized, but the old scheme for managing the base system as one unit is still in place, too. Many software packages can be installed either as binary packages or as ports, with essentially similar results but different implications for future updates. However, cross-coverage is not complete; some things can only be installed as a port or as a package.

Part of the project definition for FreeBSD 12 is to shift the system more decisively toward universal package management. The base system and ports may both continue to exist in some form (it’s currently too early to tell exactly how things will work out), but the future direction is clear.

Accordingly, try to manage add-on software with `pkg` to the extent possible. Avoid ports unless the software you want has no packagized version or you need to customize compile-time options.

Another peculiar remnant of the big-iron UNIX era is FreeBSD’s insistence that add-on packages are “local,” even though they are compiled by FreeBSD and released as part of an official package repository. Packages install binaries under `/usr/local`, and most configuration files end up in `/usr/local/etc` rather than `/etc`. 
The base system

The base system is updated as a single unit and is functionally distinct from any add-on packages (at least in theory). The base system is maintained in a Subversion repository. You can browse the source tree, including all the source branches, at svnweb.freebsd.org.

Several development branches are defined:

- The CURRENT branch is meant only for active development purposes. It is the first to receive new features and fixes but is not widely tested by the user community.

- The STABLE branch is regularly updated with improvements intended for the next major release. It includes new features but maintains package compatibility and undergoes some testing. It may introduce bugs or breaking changes and is recommended only for the adventurous.

- The RELEASE branch is forked from STABLE when a release target is achieved. It remains mostly static. The only updates to RELEASE are security fixes and fixes for serious bugs. Official ISO images derive from the RELEASE branch, and that branch is the only one recommended for use on production systems.

View your system’s current branch with `uname -r`.

```
$ uname -r
11.0-RELEASE
```

Run the `freebsd-update` command to keep your system updated with the latest packages. Fetching updates and installing them are separate operations, but you can combine the two into a single command line:

```
$ sudo freebsd-update fetch install
```

This command retrieves and installs the latest base binaries. It’s available only for the RELEASE branch; binaries are not built for the STABLE and CURRENT branches. You can use the same tool to upgrade between releases of the system. For example:

```
$ sudo freebsd-update -r 11.1-RELEASE upgrade
```
pkg: the FreeBSD package manager

pkg is intuitive and fast. It’s the easiest way to install software that isn’t already included in the base system. Use pkg help for a quick reference on the available subcommands, or pkg help command to display the man page for a particular subcommand. Table 6.2 lists some of the most frequently used subcommands.

Table 6.2: Example pkg subcommands

<table>
<thead>
<tr>
<th>Command</th>
<th>What it does</th>
</tr>
</thead>
<tbody>
<tr>
<td>pkg install -y package</td>
<td>Installs without asking any “are you sure?” questions</td>
</tr>
<tr>
<td>pkg backup</td>
<td>Makes a backup of the local package database</td>
</tr>
<tr>
<td>pkg info</td>
<td>Lists all installed packages</td>
</tr>
<tr>
<td>pkg info package</td>
<td>Shows extended information for a package</td>
</tr>
<tr>
<td>pkg search -i package</td>
<td>Searches package repository (case insensitive)</td>
</tr>
<tr>
<td>pkg audit -F</td>
<td>Shows packages with known security vulnerabilities</td>
</tr>
<tr>
<td>pkg which file</td>
<td>Shows which package owns the named file</td>
</tr>
<tr>
<td>pkg autoremove</td>
<td>Removes unused packages</td>
</tr>
<tr>
<td>pkg delete package</td>
<td>Uninstalls a package (same as remove)</td>
</tr>
<tr>
<td>pkg clean -ay</td>
<td>Removes cached packages from /var/cache/pkg</td>
</tr>
<tr>
<td>pkg update</td>
<td>Updates local copy of the package catalog</td>
</tr>
<tr>
<td>pkg upgrade</td>
<td>Upgrades packages to the latest version</td>
</tr>
</tbody>
</table>

When you install packages with pkg install, pkg consults the local package catalog, then downloads the requested package from the repository at pkg.FreeBSD.org. Once the package is installed, it’s registered in a SQLite database kept in /var/db/pkg/local.sqlite. Take care not to delete this file lest your system lose track of which packages have been installed. Create backups of the database with the pkg backup subcommand.

pkg version, a subcommand for comparing package versions, has an idiosyncratic syntax. It uses the =, <, and > characters to show packages that are current, older than the latest available version, or newer than the current version. Use the following command to list packages that have updates:

```
freebsd$ pkg version -vll

dri-11.2.2,2       < needs updating (index has 13.0.4,2)
gbm-11.2.2          < needs updating (index has 13.0.4)
harfbuzz-1.4.1     < needs updating (index has 1.4.2)
libEGL-11.2.2       < needs updating (index has 13.0.4_1)
```

This command compares all installed packages to the index (-I), looking for those that are not (-L) the current version (=), and printing verbose information (-v).
pkg search is faster than Google for finding packages. For example, pkg search dns finds all packages with “dns” in their names. The search term is a regular expression, so you can search for something like pkg search ^apache. See pkg help search for details.
The ports collection

FreeBSD ports are a collection of all the software that FreeBSD can build from source. After the ports tree is initialized, you’ll find all the available software in categorized subdirectories of /usr/ports. To initialize the ports tree, use the `portsnap` utility:

```
freenbsd$ portsnap fetch extract
```

To update the ports tree in one command, use `portsnap fetch update`.

It takes some time to download the ports metadata. The download includes pointers to the source code for all the ports, plus any associated patches for FreeBSD compatibility. When installation of the metadata is complete, you can search for software, then build and install anything you need.

For example, the `zsh` shell is not included in the FreeBSD base. Use the `whereis` utility to search for `zsh`, then build and install from the ports tree:

```
freenbsd$ whereis zsh
bash: /usr/ports/shells/zsh
freenbsd$ cd /usr/ports/shells/zsh
freenbsd$ make install clean
```

To remove software installed through the ports system, run `make deinstall` from the appropriate directory.

There’s more than one way to update ports, but we prefer the `portmaster` utility. First install `portmaster` from the ports collection:

```
freenbsd$ cd /usr/ports/ports-mgmt/portmaster
freenbsd$ make install clean
```

Run `portmaster -L` to see all the ports having updates available, and update them all at once with `portmaster -a`.

You can also install ports through the `portmaster`. In fact, it’s somewhat more convenient than the typical `make`-based process because you don’t need to leave your current directory. To install `zsh`:

```
freenbsd$ portmaster shells/zsh
```

If you need to free up some disk space, clean up the ports’ working directories with `portmaster -c`. 
6.6 Software Localization and Configuration

Adapting systems to your local (or cloud) environment is one of the prime battlegrounds of system administration. Addressing localization issues in a structured and reproducible way helps avoid the creation of snowflake systems that are impossible to recover after a major incident.

We have more to say in this book about these issues. In particular, Chapter 23, Configuration Management, and Chapter 26, Continuous Integration and Delivery, discuss tools that structure these tasks. Configuration management systems are your go-to tools for installing and configuring software in a reproducible manner. They are the master key to sane localization.

Implementation issues aside, how do you know if your local environment is properly designed? Here are a few points to consider:

- Nonadministrators should not have root privileges. Any need for root privileges in the course of normal operations is suspicious and probably indicates that something is fishy with your local configuration.

- Systems should facilitate work and not get in users’ way. Users do not wreck the system intentionally. Design internal security so that it guards against unintentional errors and the widespread dissemination of administrative privileges.

- Misbehaving users are learning opportunities. Interview them before you chastise them for not following proper procedures. Users frequently respond to inefficient administrative procedures by working around them, so always consider the possibility that noncompliance is an indication of architectural problems.

- Be customer-centered. Talk to users and ask them which tasks they find difficult in your current configuration. Find ways to make these tasks simpler.

- Your personal preferences are yours. Let your users have their own. Offer choices wherever possible.

- When administrative decisions affect users’ experience of the system, be aware of the reasons for your decisions. Let your reasons be known.

- Keep your local documentation up to date and easily accessible. See this page for more information on this topic.
Organizing your localization

If your site has a thousand computers and each computer has its own configuration, you will spend a major portion of your working time figuring out why one box has a particular problem and another doesn’t. Clearly, the solution is to make every computer the same…right? But real-world constraints and the varying needs of users typically make this solution impossible.

There’s a big difference in administrability between *multiple* configurations and *countless* configurations. The trick is to split your setup into manageable bits. Some parts of the localization apply to all managed hosts, others apply to only a few, and still others are specific to individual boxes. Even with the convenience of configuration management tools, try not to allow too much drift among systems.

However you design your localization system, make sure that all original data is kept in a revision control system. This precaution lets you keep track of which changes have been thoroughly tested and are ready for deployment. In addition, it lets you identify the originator of any problematic changes. The more people involved in the process, the more important this last consideration becomes.
Structuring updates

In addition to performing initial installations, you will also need to continually roll out updates. This remains one of the most important security tasks. Keep in mind, though, that different hosts have different needs for concurrency, stability, and uptime.

Do not roll out new software releases en masse. Instead, stage rollouts according to a gradual plan that accommodates other groups’ needs and allows time for problems to be discovered while their potential to cause damage is still limited. This sometimes referred to as a “canary” release process, named for the fabled “canary in the coal mine.” In addition, never update critical servers until you have some confidence in the changes you are contemplating. Avoid rolling out changes on Fridays unless you’re prepared for a long weekend in front of the terminal.

It’s usually advantageous to separate the base OS release from the localization release. Depending on the stability needs of your environment, you might choose to use minor local releases only for bug fixing. However, we have found that adding new features in small doses yields a smoother operation than queuing up changes into “horse pill” releases that risk a major disruption of service. This principle is closely related to the idea of continuous integration and deployment; see Chapter 26.
Limiting the field of play

It’s often a good idea to specify a maximum number of “releases” you are willing to have in play at any given time. Some administrators see no reason to fix software that isn’t broken. They point out that gratuitously upgrading systems costs time and money and that “cutting edge” all too often means “bleeding edge.” Those who put these principles into practice must be willing to collect an extensive catalog of active releases.

By contrast, the “lean and mean” crowd point to the inherent complexity of releases and the difficulty of comprehending (let alone managing) a random collection of releases dating years into the past. Their trump cards are security patches, which must typically be applied universally and on a strict schedule. Patching outdated versions of the operating system is often infeasible, so administrators are faced with the choice of skipping updates on some computers or crash-upgrading these machines to a newer internal release. Not good.

Neither of these perspectives is provably correct, but we tend to side with those who favor a limited number of releases. Better to perform your upgrades on your own schedule rather than one dictated by an external emergency.
Testing

It’s important to test changes before unleashing them on the world. At a minimum, this means that you need to test your own local configuration changes. However, you should really test the software that your vendor releases as well. A major UNIX vendor once released a patch that performed an `rm -rf /`. Imagine installing this patch throughout your organization without testing it first.

Testing is an especially pertinent issue if you use a service that offers an automatic patching capability, such as most of the packaging systems discussed in this chapter. Never connect mission-critical systems directly to a vendor-sponsored update service. Instead, point most of your systems to an internal mirror that you control, and test updates on noncritical systems first.

See this page for more information about trouble tracking.

If you foresee that an update might cause user-visible problems or changes, notify users well in advance and give them a chance to communicate with you if they have concerns regarding your intended changes or timing. Make sure that users have an easy way to report bugs.

If your organization is geographically distributed, make sure that other offices help with testing. International participation is particularly valuable in multilingual environments. If no one in the U.S. office speaks Japanese, for example, you had better get the Tokyo office to test anything that might affect Unicode support. A surprising number of system parameters vary with location. Does the new version of software you’re installing break UTF-8 encoding, rendering text illegible for some languages?
6.7 RECOMMENDED READING


LAWSON, NOLAN. *What it feels like to be an open-source maintainer.* wp.me/p1t8Ca-1ry

PXELinux Questions. syslinux.zytor.com/wiki/index.php/PXELINUX

RODIN, JOSIP. *Debian New Maintainers’ Guide.* debian.org/doc/maint-guide
This document contains good information about .deb packages. See also Chapter 7 of the Debian FAQ and Chapter 2 of the Debian reference manual.
A scalable approach to system management requires that administrative changes be structured, reproducible, and replicable across multiple computers. In the real world, that means those changes should be mediated by software rather than performed by administrators working from checklists—or worse, from memory.

Scripts standardize administrative chores and free up admins’ time for more important and more interesting tasks. Scripts also serve as a kind of low-rent documentation in that they record the steps needed to complete a particular task.

Sysadmins’ main alternative to scripting is to use the configuration management systems described in Chapter 23. These systems offer a structured approach to administration that scales well to the cloud and to networks of machines. However, they are more complex, more formal, and less flexible than plain-vanilla scripting. In practice, most administrators use a combination of scripting and configuration management. Each approach has its strengths, and they work well together.

This chapter takes a quick look at sh, Python, and Ruby as languages for scripting. We cover some basic tips for using the shell and also discuss regular expressions as a general technology.
7.1 Scripting Philosophy

This chapter includes a variety of scripting tidbits and language particulars. That information is useful, but more important than any of those details is the broader question of how to incorporate scripting (or more generally, automation) into your mental model of system administration.
Write microscripts

New sysadmins often wait to learn scripting until they’re confronted with a particularly complex or tedious chore. For example, maybe it’s necessary to automate a particular type of backup so that it’s done regularly and so that the backup data is stored in two different data centers. Or perhaps there’s a cloud server configuration that would be helpful to create, initialize, and deploy with a single command.

These are perfectly legitimate scripting projects, but they can leave the impression that scripting is an elephant gun to be unboxed only when big game is on the horizon. After all, that first 100-line script probably took several days to write and debug. You can’t be spending days on every little task…can you?

Actually, you achieve most efficiencies by saving a few keystrokes here and a few commands there. Marquee-level scripts that are part of your site’s formal procedures are just the visible portion of a much larger iceberg. Below the waterline lie many smaller forms of automation that are equally useful for sysadmins. As a general rule, approach every chore with the question, “How can I avoid having to deal with this issue again in the future?”

Most admins keep a selection of short scripts for personal use (aka scriptlets) in their ~/bin directories. Use these quick-and-dirty scripts to address the pain points you encounter in day-to-day work. They are usually short enough to read at a glance, so they don’t need documentation beyond a simple usage message. Keep them updated as your needs change.

For shell scripts, you also have the option of defining functions that live inside your shell configuration files (e.g., .bash_profile) rather than in freestanding script files. Shell functions work similarly to stand-alone scripts, but they are independent of your search path and automatically travel with you wherever you take your shell environment.

Just as a quick illustration, here’s a simple Bash function that backs up files according to a standardized naming convention:

```bash
function backup () {
    newname=$1.`date +%Y-%m-%d.%H%M.bak`
    mv $1 $newname;
    echo "Backed up $1 to $newname."
    cp -p $newname $1;
}
```

Despite the function-like syntax, you use it just like a script or any other command:

```bash
$ backup afile
Backed up afile to afile.2017-02-05.1454.bak.
```

The main disadvantage of shell functions is that they’re stored in memory and have to be reparsed every time you start a new shell. But on modern hardware, these costs are negligible.
At a smaller scale still are aliases, which are really just an extra-short variety of scriptlet. These can be defined either with shell functions or with your shell’s built-in aliasing feature (usually called alias). Most commonly, they set default arguments for individual commands. For example,

```
alias ls='ls -Fh'
```

makes the `ls` command punctuate the names of directories and executables and requests human-readable file sizes for long listings (e.g., 2.4M).
Learn a few tools well

System administrators encounter a lot of software. They can’t be experts at everything, so they usually become skilled at skimming documentation, running experiments, and learning just enough about new software packages to configure them for the local environment. Laziness is a virtue.

That said, some topics are valuable to study in detail because they amplify your power and effectiveness. In particular, you should know a shell, a text editor, and a scripting language thoroughly. (Not to spoil the rest of this chapter, but these should probably be Bash, vim, and Python.) Read the manuals from front to back, then regularly read books and blogs. There’s always more to learn.

Enabling technologies like these reward up-front study for a couple of reasons. As tools, they are fairly abstract; it’s hard to envision all the things they’re capable of doing without reading about the details. You can’t use features you’re not aware of.

Another reason these tools reward exploration is that they’re “made of meat”; most features are potentially valuable to most administrators. Compare that with the average server daemon, where your main challenge is often to identify the 80% of features that are irrelevant to your situation.

A shell or editor is a tool you use constantly. Every incremental improvement in your proficiency with these tools translates not only into increased productivity but also into greater enjoyment of the work. No one likes to waste time on repetitive details.
Automate all the things

Shell scripts aren’t system administrators’ only opportunity to benefit from automation. There’s a whole world of programmable systems out there—just keep an eye out for them. Exploit these facilities aggressively and use them to impedance-match your tools to your workflow.

For example, we created this book in Adobe InDesign, which is ostensibly a GUI application. However, it’s also scriptable in JavaScript, so we created a library of InDesign scripts to implement and enforce many of our conventions.

Such opportunities are everywhere:

- Microsoft Office apps are programmable in Visual Basic or C#. If your work involves analysis or reporting, make those TPS reports write themselves.

- Most Adobe applications are scriptable.

- If your responsibilities include database wrangling, you can automate many routine tasks with SQL stored procedures. Some databases even support additional languages; for example, PostgreSQL speaks Python.

- PowerShell is the mainstream scripting tool for Microsoft Windows systems. Third party add-ons like AutoHotKey go a long way toward facilitating the automation of Windows apps.

- On macOS systems, some applications can be controlled through AppleScript. At the system level, use the Automator app, the Services system, and folder actions to automate various chores and to connect traditional scripting languages to the GUI.

Within the world of system administration specifically, a few subsystems have their own approaches to automation. Many others play well with general-purpose automation systems such as Ansible, Salt, Chef, and Puppet, described in Chapter 23, Configuration Management. For everything else, there’s general-purpose scripting.
Don’t optimize prematurely

There’s no real distinction between “scripting” and “programming.” Language developers sometimes take offense when their babies are lumped into the “scripting” category, not just because the label suggests a certain lack of completeness, but also because some scripting languages of the past have earned reputations for poor design.

We still like the term “scripting,” though; it evokes the use of software as a kind of universal glue that binds various commands, libraries, and configuration files into a more functional whole.

Administrative scripts should emphasize programmer efficiency and code clarity rather than computational efficiency. This is not an excuse to be sloppy, but simply a recognition that it rarely matters whether a script runs in half a second or two seconds. Optimization can have an amazingly low return on investment, even for scripts that run regularly out of cron.
Pick the right scripting language

For a long time, the standard language for administrative scripts was the one defined by the `sh` shell. Shell scripts are typically used for light tasks such as automating a sequence of commands or assembling several filters to process data.

The shell is always available, so shell scripts are relatively portable and have few dependencies other than the commands they invoke. Whether or not you choose the shell, the shell might choose you: most environments include a hefty complement of existing `sh` scripts, and administrators frequently need to read, understand, and tweak those scripts.

As a programming language, `sh` is somewhat inelegant. The syntax is idiosyncratic, and the shell lacks the advanced text processing features of modern languages—features that are often of particular use to system administrators.

Perl, designed in the late 1980s, was a major step forward for script-writing administrators. Its permissive syntax, extensive library of user-written modules, and built-in support of regular expressions made it an administrative favorite for many years. Perl permits (and some would say, encourages) a certain “get it done and damn the torpedoes” style of coding. Opinions differ on whether that’s an advantage or a drawback.

These days, Perl is known as Perl 5 to distinguish it from the redesigned and incompatible Perl 6, which has finally reached general release after 15 years of gestation. Unfortunately, Perl 5 is showing its age in comparison with newer languages, and use of Perl 6 isn’t yet widespread enough for us to recommend it as a safe choice. It might be that the world has moved on from Perl entirely. We suggest avoiding Perl for new work at this point.

JavaScript and PHP are best known as languages for web development, but they can be arm-twisted into service as general-purpose scripting tools, too. Unfortunately, both languages have design flaws that limit their appeal, and they lack many of the third party libraries that system administrators rely on.

If you come from the web development world, you might be tempted to apply your existing PHP or JavaScript skills to system administration. We recommend against this. Code is code, but living in the same ecosystem as other sysadmins brings a variety of long-term benefits. (At the very least, avoiding PHP means you won’t have to endure the ridicule of your local sysadmin Meetup.)

Python and Ruby are modern, general-purpose programming languages that are both well suited for administrative work. These languages incorporate a couple of decades’ worth of language design advancements relative to the shell, and their text processing facilities are so powerful that `sh` can only weep and cower in shame.

The main drawback to both Python and Ruby is that their environments can be a bit fussy to set up, especially when you start to use third party libraries that have compiled components written
in C. The shell skirts this particular issue by having no module structure and no third party libraries.

In the absence of outside constraints, Python is the most broadly useful scripting language for system administrators. It’s well designed, widely used, and widely supported by other packages. Table 7.1 shows some general notes on other languages.

### Table 7.1: Scripting language cheat sheet

<table>
<thead>
<tr>
<th>Language</th>
<th>Designer</th>
<th>When to use it</th>
</tr>
</thead>
<tbody>
<tr>
<td>Bourne shell</td>
<td>Stephen Bourne</td>
<td>Simple series of commands, portable scripts</td>
</tr>
<tr>
<td>bash</td>
<td>Brian Fox</td>
<td>Like Bourne shell; nicer but less portable</td>
</tr>
<tr>
<td>C shell</td>
<td>Bill Joy</td>
<td>Never for scripting</td>
</tr>
<tr>
<td>JavaScript</td>
<td>Brendan Eich</td>
<td>Web development, app scripting</td>
</tr>
<tr>
<td>Perl</td>
<td>Larry Wall</td>
<td>Quick hacks, one-liners, text processing</td>
</tr>
<tr>
<td>PHP</td>
<td>Rasmus Lerdorf</td>
<td>You’ve been bad and deserve punishment</td>
</tr>
<tr>
<td>Python</td>
<td>Guido van Rossum</td>
<td>General-purpose scripting, data wrangling</td>
</tr>
<tr>
<td>Ruby</td>
<td>“Matz” Matsumoto</td>
<td>General-purpose scripting, web</td>
</tr>
</tbody>
</table>
Follow best practices

Although the code fragments in this chapter contain few comments and seldom print usage messages, that’s only because we’ve skeletonized each example to make specific points. Real scripts should behave better. There are whole books on best practices for coding, but here are a few basic guidelines:

- When run with inappropriate arguments, scripts should print a usage message and exit. For extra credit, implement --help this way, too.

- Validate inputs and sanity-check derived values. Before doing an `rm -rf` on a calculated path, for example, you might have the script double-check that the path conforms to the pattern you expect.

- Return a meaningful exit code: zero for success and nonzero for failure. You needn’t necessarily give every failure mode a unique exit code, however; consider what callers will actually want to know.

- Use appropriate naming conventions for variables, scripts, and routines. Conform to the conventions of the language, the rest of your site’s code base, and most importantly, the other variables and functions defined in the current project. Use mixed case or underscores to make long names readable. The naming of the scripts themselves is important, too. In this context, dashes are more common than underscores for simulating spaces, as in `system-config-printer`.

- Assign variable names that reflect the values they store, but keep them short. `number_of_lines_of_input` is way too long; try `n_lines`.

- Consider developing a style guide so you and your colleagues can write code according to the same conventions. A guide makes it easier for you to read other people’s code and for them to read yours. (On the other hand, style guide construction can absorb a contentious team’s attention for weeks. Don’t fight over the style guide; cover the areas of agreement and avoid long negotiations over the placement of braces and commas. The main thing is to make sure everyone’s on board with a consistent set of naming conventions.)

- Start every script with a comment block that tells what the script does and what parameters it takes. Include your name and the date. If the script requires nonstandard tools, libraries, or modules to be installed on the system, list those as well.

- Comment at the level you yourself will find helpful when you return to the script after a month or two. Some useful points to comment on are the following: choices of algorithm, web references used, reasons for not doing things in a more obvious way, unusual paths through the code, anything that was a problem during development.
• Don’t clutter code with useless comments; assume intelligence and language proficiency on the part of the reader.

• It’s OK to run scripts as root, but avoid making them setuid; it’s tricky to make setuid scripts completely secure. Use `sudo` to implement appropriate access control policies instead.

• Don’t script what you don’t understand. Administrators often view scripts as authoritative documentation of how a particular procedure should be handled. Don’t set a misleading example by distributing half-baked scripts.

• Feel free to adapt code from existing scripts for your own needs. But don’t engage in “copy, paste, and pray” programming when you don’t understand the code. Take the time to figure it out. This time is never wasted.

• With `bash`, use `-x` to echo commands before they are executed and `-n` to check commands for syntax without executing them.

• Remember that in Python, you are in debug mode unless you explicitly turn it off with a `-O` argument on the command line. You can test the special `__debug__` variable before printing diagnostic output.

Tom Christiansen suggests the following five golden rules for producing useful error messages:

• Error messages should go to STDERR, not STDOUT (see this page).

• Include the name of the program that’s issuing the error.

• State what function or operation failed.

• If a system call fails, include the `perror` string.

• Exit with some code other than 0.
7.2 SHELL BASICS

UNIX has always offered users a choice of shells, but some version of the Bourne shell, \texttt{sh}, has been standard on every UNIX and Linux system. The code for the original Bourne shell never made it out of AT&T licensing limbo, so these days \texttt{sh} is most commonly manifested in the form of the Almquist shell (known as \texttt{ash}, \texttt{dash}, or simply \texttt{sh}) or the “Bourne-again” shell, \texttt{bash}.

The Almquist shell is a reimplementation of the original Bourne shell without extra frills. By modern standards, it’s barely usable as a login shell. It exists only to run \texttt{sh} scripts efficiently.

\texttt{bash} focuses on interactive usability. Over the years, it has absorbed most of the useful features pioneered by other shells. It still runs scripts designed for the original Bourne shell, but it’s not particularly tuned for scripting. Some systems (e.g., the Debian lineage) include both \texttt{bash} and \texttt{dash}. Others rely on \texttt{bash} for both scripting and interactive use.

The Bourne shell has various other offshoots, notably \texttt{ksh} (the Korn shell) and \texttt{ksh}’s souped-up cousin \texttt{zsh}. \texttt{zsh} features broad compatibility with \texttt{sh}, \texttt{ksh}, and \texttt{bash}, as well as many interesting features of its own, including spelling correction and enhanced globbing. It’s not used as any system’s default shell (as far as we are aware), but it does have something of a cult following.

Historically, BSD-derived systems favored the C shell, \texttt{csh}, as an interactive shell. It’s now most commonly seen in an enhanced version called \texttt{tcsh}. Despite the formerly widespread use of \texttt{csh} as a login shell, it is not recommended for use as a scripting language. For a detailed explanation of why this is so, see Tom Christiansen’s classic rant, “Csh Programming Considered Harmful.” It’s widely reproduced on the web. One copy is harmful.cat-v.org/software/csh.

\texttt{tcsh} is a fine and widely available shell, but it’s not an \texttt{sh} derivative. Shells are complex; unless you’re a shell connoisseur, there’s not much value in learning one shell for scripting and a second one—with different features and syntax—for daily use. Stick to a modern version of \texttt{sh} and let it do double duty.

Among the \texttt{sh} options, \texttt{bash} is pretty much the universal standard these days. To move effortlessly among different systems, standardize your personal environment on \texttt{bash}.

FreeBSD retains \texttt{tcsh} as root’s default and does not ship \texttt{bash} as part of the base system. But that’s easily fixed: run \texttt{sudo pkg install bash} to install \texttt{bash}, and use \texttt{chsh} to change your shell or the shell of another user. You can set \texttt{bash} as the default for new users by running \texttt{adduser -C}. (Changing the default might seem presumptuous, but standard FreeBSD relegates new users to the Almquist \texttt{sh}. There’s nowhere to go but up.)

Before taking up the details of shell scripting, we should review some of the basic features and syntax of the shell.

The material in this section applies to the major interactive shells in the \texttt{sh} lineage (including \texttt{bash} and \texttt{ksh}, but not \texttt{csh} or \texttt{tcsh}), regardless of the exact platform you are using. Try out the
forms you’re not familiar with and experiment!
Command editing

We’ve watched too many people edit command lines with the arrow keys. You wouldn’t do that in your text editor, right?

If you like **emacs**, all the basic **emacs** commands are available to you when you’re editing history. <Control-E> goes to the end of the line and <Control-A> to the beginning. <Control-P> steps backward through recently executed commands and recalls them for editing. <Control-R> searches incrementally through your history to find old commands.

If you like **vi/vim**, put your shell’s command-line editing into **vi** mode like this:

```
$ set -o vi
```

As in **vi**, editing is modal; however, you start in input mode. Press <Esc> to leave input mode and “i” to reenter it. In edit mode, “w” takes you forward a word, “fX” finds the next X in the line, and so on. You can walk through past command history entries with <Esc> k. Want **emacs** editing mode back again?

```
$ set -o emacs
```
Pipes and redirection

Every process has at least three communication channels available to it: standard input (STDIN), standard output (STDOUT), and standard error (STDERR). Processes initially inherit these channels from their parents, so they don’t necessarily know where they lead. They might connect to a terminal window, a file, a network connection, or a channel belonging to another process, to name a few possibilities.

UNIX and Linux have a unified I/O model in which each channel is named with a small integer called a file descriptor. The exact number assigned to a channel is not usually significant, but STDIN, STDOUT, and STDERR are guaranteed to correspond to file descriptors 0, 1, and 2, so it’s safe to refer to these channels by number. In the context of an interactive terminal window, STDIN normally reads from the keyboard and both STDOUT and STDERR write their output to the screen.

Many traditional UNIX commands accept their input from STDIN and write their output to STDOUT. They write error messages to STDERR. This convention lets you string commands together like building blocks to create composite pipelines.

The shell interprets the symbols <, >, and >> as instructions to reroute a command’s input or output to or from a file. A < symbol connects the command’s STDIN to the contents of an existing file. The > and >> symbols redirect STDOUT; > replaces the file’s existing contents, and >> appends to them. For example, the command

```
$ grep bash /etc/passwd > /tmp/bash-users
```

copies lines containing the word “bash” from /etc/passwd to /tmp/bash-users, creating the file if necessary. The command below sorts the contents of that file and prints them to the terminal.

```
$ sort < /tmp/bash-users
root:x:0:0:root:/root:/bin/bash
...
```

Truth be told, the sort command accepts filenames, so the < symbol is optional in this context. It’s used here for illustration.

To redirect both STDOUT and STDERR to the same place, use the >& symbol. To redirect STDERR only, use 2>.

The find command illustrates why you might want separate handling for STDOUT and STDERR because it tends to produce output on both channels, especially when run as an unprivileged user. For example, a command such as

```
$ find / -name core
```
usually results in so many “permission denied” error messages that genuine hits get lost in the clutter. To discard all the error messages, use

```
$ find / -name core 2> /dev/null
```

In this version, only real matches (where the user has read permission on the parent directory) come to the terminal window. To save the list of matching paths to a file, use

```
$ find / -name core > /tmp/corefiles 2> /dev/null
```

This command line sends matching paths to `/tmp/corefiles`, discards errors, and sends nothing to the terminal window.

To connect the STDOUT of one command to the STDIN of another, use the `|` symbol, commonly known as a pipe. For example:

```
$ find / -name core 2> /dev/null | less
```

The first command runs the same `find` operation as the previous example, but sends the list of discovered files to the `less` pager rather than to a file. Another example:

```
$ ps -ef | grep httpd
```

This one runs `ps` to generate a list of processes and pipes it to the `grep` command, which selects lines that contain the word `httpd`. The output of `grep` is not redirected, so the matching lines come to the terminal window.

```
$ cut -d: -f7 < /etc/passwd | sort -u
```

Here, the `cut` command picks out the path to each user’s shell from `/etc/passwd`. The list of shells is then sent through `sort -u` to produce a sorted list of unique values.

To execute a second command only if its precursor completes successfully, you can separate the commands with an `&&` symbol. For example,

```
$ mkdir foo && cd foo
```

attempts to create a directory called `foo`, and if the directory was successfully created, executes `cd`. Here, the success of the `mkdir` command is defined as its yielding an exit code of zero, so the use of a symbol that suggests “logical AND” for this purpose might be confusing if you’re accustomed to short-circuit evaluation in other programming languages. Don’t think about it too much; just accept it as a shell idiom.

Conversely, the `||` symbol executes the following command only if the preceding command fails (that is, it produces a nonzero exit status). For example,

```
$ cd foo || echo "No such directory"
```
In a script, you can use a backslash to break a command onto multiple lines. This feature can help to distinguish error-handling code from the rest of a command pipeline:

```bash
cp --preserve --recursive /etc/* /spare/backup \    || echo "Did NOT make backup"
```

For the opposite effect—multiple commands combined onto one line—you can use a semicolon as a statement separator:

```bash
$ mkdir foo; cd foo; touch afile
```
Variables and quoting

Variable names are unmarked in assignments but prefixed with a dollar sign when their values are referenced. For example:

```
$ etcdir='/etc'
$ echo $etcdir
/etc
```

Omit spaces around the = symbol; otherwise, the shell mistakes your variable name for a command name and treats the rest of the line as a series of arguments to that command.

When referencing a variable, you can surround its name with curly braces to clarify to the parser and to human readers where the variable name stops and other text begins; for example, `${etcdir}` instead of just `$etcdir`. The braces are not normally required, but they can be useful when you want to expand variables inside double-quoted strings. Often, you’ll want the contents of a variable to be followed by literal letters or punctuation. For example,

```
$ echo "Saved ${rev}th version of mdadm.conf."
Saved 8th version of mdadm.conf.
```

There’s no standard convention for the naming of shell variables, but all-caps names typically suggest environment variables or variables read from global configuration files. More often than not, local variables are all-lowercase with components separated by underscores. Variable names are case sensitive.

The shell treats strings enclosed in single and double quotes similarly, except that double-quoted strings are subject to globbing (the expansion of filename-matching metacharacters such as * and ?) and variable expansion. For example:

```
$ mylang="Pennsylvania Dutch"
$ echo "I speak ${mylang}." 
I speak Pennsylvania Dutch.
$ echo 'I speak ${mylang}.'
I speak ${mylang}.
```

Backquotes, also known as backticks, are treated similarly to double quotes, but they have the additional effect of executing the contents of the string as a shell command and replacing the string with the command’s output. For example,

```
$ echo "There are `wc -l < /etc/passwd` lines in the passwd file."
There are 28 lines in the passwd file.
```
Environment variables

When a UNIX process starts up, it receives a list of command-line arguments and also a set of “environment variables.” Most shells show you the current environment in response to the `printenv` command:

```
$ printenv
EDITOR=vi
USER=garth
ENV=/home/garth/.bashrc
LSCOLORS=exfxgxgdgxgxgbxhxchx
PWD=/mega/Documents/Projects/Code/spl
HOME=/home/garth
... <total of about 50>
```

By convention, environment variables have all-caps names, but that is not technically required.

Programs that you run can consult these variables and change their behavior accordingly. For example, `vipw` checks the `EDITOR` environment variable to determine which text editor to run for you.

Environment variables are automatically imported into sh’s variable namespace, so they can be set and read with the standard syntax. Use `export varname` to promote a shell variable to an environment variable. You can also combine this syntax with a value assignment, as seen here:

```
$ export EDITOR=nano
$ vipw
<starts the nano editor>
```

Despite being called “environment” variables, these values don’t exist in some abstract, ethereal place outside of space and time. The shell passes a snapshot of the current values to any program you run, but no ongoing connection exists. Moreover, every shell or program—and every terminal window—has its own distinct copy of the environment that can be separately modified.

Commands for environment variables that you want to set up at login time should be included in your `~/.profile` or `~/.bash_profile` file. Other environment variables, such as `PWD` for the current working directory, are automatically maintained by the shell.
**Common filter commands**

Any well-behaved command that reads STDIN and writes STDOUT can be used as a filter (that is, a component of a pipeline) to process data. In this section we briefly review some of the more widely used filter commands (including some used in passing above), but the list is practically endless. Filter commands are so team oriented that it’s sometimes hard to show their use in isolation.

Most filter commands accept one or more filenames on the command line. Only if you do not specify a file do they read their standard input.

**cut: separate lines into fields**

The **cut** command prints selected portions of its input lines. It most commonly extracts delimited fields, as in the example on this page, but it can return segments defined by column boundaries as well. The default delimiter is <Tab>, but you can change delimiters with the -d option. The -f options specifies which fields to include in the output.

For an example of the use of **cut**, see the section on **uniq**, below.

**sort: sort lines**

**sort** sorts its input lines. Simple, right? Well, maybe not—there are a few potential subtleties regarding the exact parts of each line that are sorted (the “keys”) and the collation order to be imposed. **Table 7.2** shows a few of the more common options, but check the man page for others.

**Table 7.2: sort options**

<table>
<thead>
<tr>
<th>Opt</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>-b</td>
<td>Ignore leading whitespace</td>
</tr>
<tr>
<td>-f</td>
<td>Sort case-insensitively</td>
</tr>
<tr>
<td>-h</td>
<td>Sort “human readable” numbers (e.g., 2MB)</td>
</tr>
<tr>
<td>-k</td>
<td>Specify the columns that form the sort key</td>
</tr>
<tr>
<td>-n</td>
<td>Compare fields as integer numbers</td>
</tr>
<tr>
<td>-r</td>
<td>Reverse sort order</td>
</tr>
<tr>
<td>-t</td>
<td>Set field separator (the default is whitespace)</td>
</tr>
<tr>
<td>-u</td>
<td>Output only unique records</td>
</tr>
</tbody>
</table>

The commands below illustrate the difference between numeric and dictionary sorting, which is the default. Both commands use the **-t**: and **-k3,3** options to sort the **/etc/group** file by its third colon-separated field, the group ID. The first sorts numerically and the second alphabetically.
sort accepts the key specification -k3 (rather than -k3,3), but it probably doesn’t do what you expect. Without the terminating field number, the sort key continues to the end of the line.

Also useful is the -h option, which implements a numeric sort that understands suffixes such as M for mega and G for giga. For example, the following command correctly sorts the sizes of directories under /usr while maintaining the legibility of the output:

`$ du -sh /usr/* | sort -h`

16K  /usr/locale
128K  /usr/local
648K  /usr/games
15M  /usr/sbin
20M  /usr/include
117M  /usr/src
126M  /usr/bin
845M  /usr/share
1.7G  /usr/lib

uniq: print unique lines

uniq is similar in spirit to sort -u, but it has some useful options that sort does not emulate: -c to count the number of instances of each line, -d to show only duplicated lines, and -u to show only nonduplicated lines. The input must be presorted, usually by being run through sort.

For example, the command below shows that 20 users have /bin/bash as their login shell and that 12 have /bin/false. (The latter are either pseudo-users or users whose accounts have been disabled.)

`$ cut -d: -f7 /etc/passwd | sort | uniq -c`

20 /bin/bash
12 /bin/false

wc: count lines, words, and characters

Counting the number of lines, words, and characters in a file is another common operation, and the wc (word count) command is a convenient way of doing this. Run without options, it displays
In the context of scripting, it is more common to supply a `-l`, `-w`, or `-c` option to make `wc`'s output consist of a single number. This form is most commonly seen inside backquotes so that the result can be saved or acted on.

**tee: copy input to two places**

A command pipeline is typically linear, but it’s often helpful to tap into the data stream and send a copy to a file or to the terminal window. You can do this with the `tee` command, which sends its standard input both to standard out and to a file that you specify on the command line. Think of it as a tee fixture in plumbing.

The device `/dev/tty` is a synonym for the current terminal window. For example,

```
$ find / -name core | tee /dev/tty | wc -l
```

prints both the pathnames of files named `core` and a count of the number of `core` files that were found.

A common idiom is to terminate a pipeline that will take a long time to run with a `tee` command. That way, output goes both to a file and to the terminal window for inspection. You can preview the initial results to make sure everything is working as you expected, then leave while the command runs, knowing that the results will be saved.

**head and tail: read the beginning or end of a file**

Reviewing lines from the beginning or end of a file is a common administrative operation. These commands display ten lines of content by default, but you can use the `-n numlines` option to specify more or fewer.

For interactive use, `head` is more or less obsoleted by the `less` command, which paginates files for display. But `head` still finds plenty of use within scripts.

`tail` also has a nifty `-f` option that’s particularly useful for sysadmins. Instead of exiting immediately after printing the requested number of lines, `tail -f` waits for new lines to be added to the end of the file and prints them as they appear—great for monitoring log files. Be aware, however, that the program writing the file might be buffering its own output. Even if lines are being added at regular intervals from a logical perspective, they might only become visible in chunks of 1KiB or 4KiB.

`head` and `tail` accept multiple filenames on the command line. Even `tail -f` allows multiple files, and this feature can be quite handy; when new output appears, `tail` prints the name of the file in which it appeared.
Type <Control-C> to stop monitoring.

**grep: search text**

grep searches its input text and prints the lines that match a given pattern. Its name derives from the `g/regular-expression/p` command in the ed editor, which came with the earliest versions of UNIX (and is still present on current systems).

“Regular expressions” are text-matching patterns written in a standard and well-characterized pattern-matching language. They’re a universal standard used by most programs that do pattern matching, although there are minor variations among implementations. The odd name stems from regular expressions’ origins in theory-of-computation studies. We discuss regular expression syntax in more detail starting [here](#).

Like most filters, grep has many options, including `-c` to print a count of matching lines, `-i` to ignore case when matching, and `-v` to print nonmatching (rather than matching) lines. Another useful option is `-l` (lower case L), which makes grep print only the names of matching files rather than printing each line that matches. For example, the command

```
$ sudo grep -l mdadm /var/log/*
/var/log/auth.log
/var/log/syslog.0
```

shows that log entries from mdadm have appeared in two different log files.

grep is traditionally a fairly basic regular expression engine, but some versions permit the selection of other dialects. For example, `grep -P` on Linux selects Perl-style expressions, though the man page warns darkly that they are “highly experimental.” If you need full power, just use Ruby, Python, or Perl.

If you filter the output of `tail -f` with grep, add the `--line-buffered` option to make sure you see each matching line as soon as it becomes available:

```
$ tail -f /var/log/messages | grep --line-buffered ZFS
May 8 00:44:00 nutrient ZFS: vdev state changed, pool_
    guid=10151087465118396807 vdev_guid=7163376375690181882
...
7.3 SH SCRIPTING

sh is great for simple scripts that automate things you’d otherwise be typing on the command line. Your command-line skills carry over to sh scripting, and vice versa, which helps you extract maximum value from the learning time you invest in sh derivatives. But once an sh script gets above 50 lines or when you need features that sh doesn’t have, it’s time to move on to Python or Ruby.

For scripting, there’s some value in limiting yourself to the dialect understood by the original Bourne shell, which is both an IEEE and a POSIX standard. sh-compatible shells often supplement this baseline with additional language features. It’s fine to use these extensions if you do so deliberately and are willing to require a specific interpreter. But more commonly, scripters end up using these extensions inadvertently and are then surprised when their scripts don’t run on other systems.

In particular, don’t assume that the system’s version of sh is always bash, or even that bash is available. Ubuntu replaced bash with dash as the default script interpreter in 2006, and as part of that conversion process they compiled a handy list of bashisms to watch out for. You can find it at wiki.ubuntu.com/DashAsBinSh.
Execution

Sh comments start with a sharp (#) and continue to the end of the line. As on the command line, you can break a single logical line onto multiple physical lines by escaping the newline with a backslash. You can also put more than one statement on a line by separating the statements with semicolons.

An sh script can consist of nothing but a series of command lines. For example, the following helloworld script simply does an echo.

```
#!/bin/sh
echo "Hello, world!"
```

The first line is known as the “shebang” statement and declares the text file to be a script for interpretation by /bin/sh (which might itself be a link to dash or bash). The kernel looks for this syntax when deciding how to execute the file. From the perspective of the shell spawned to execute the script, the shebang line is just a comment.

In theory, you would need to adjust the shebang line if your system’s sh was in a different location. However, so many existing scripts assume /bin/sh that systems are compelled to support it, if only through a link.

If you need your script to run under bash or another interpreter that might not have the same command path on every system, you can use /usr/bin/env to search your PATH environment variable for a particular command. For example,

```
#!/usr/bin/env ruby
```

is a common idiom for starting Ruby scripts. Like /bin/sh, /usr/bin/env is such a widely-relied-on path that all systems are obliged to support it.

Path searching has security implications, particularly when running scripts under sudo. See this page for more information about sudo’s handling of environment variables.

To prepare a script for running, just turn on its execute bit (see this page).

```
$ chmod +x helloworld
$ ./helloworld
Hello, world!
```

If your shell understands the command helloworld without the ./ prefix, that means the current directory (.) is in your search path. This is bad because it gives other users the opportunity to lay traps for you in the hope that you’ll try to execute certain commands while cd’ed to a directory on which they have write access.

You can also invoke the shell as an interpreter directly:
$ sh helloworld
Hello, world!

$ source helloworld
Hello, world!

The first command runs **helloworld** in a new instance of **sh**, and the second makes your existing login shell read and execute the contents of the file. The latter option is useful when the script sets up environment variables or makes other customizations that apply only to the current shell. It’s commonly used in scripting to incorporate the contents of a configuration file written as a series of variable assignments. The “dot” command is a synonym for **source**, e.g., `. helloworld`.  

---

See [this page](#) for more information about permission bits.

---

If you come from the Windows world, you might be accustomed to a file’s extension indicating what type of file it is and whether it can be executed. In UNIX and Linux, the file permission bits determine whether a file can be executed, and if so, by whom. If you wish, you can give your shell scripts a `.sh` suffix to remind you what they are, but you’ll then have to type out the `.sh` when you run the command, since UNIX doesn’t treat extensions specially.
From commands to scripts

Before we jump into sh’s scripting features, a note about methodology. Most people write sh scripts the same way they write Python or Ruby scripts: with a text editor. However, it’s more productive to think of your regular shell command prompt as an interactive script development environment.

For example, suppose you have log files named with the suffixes .log and .LOG scattered throughout a directory hierarchy and that you want to change them all to the uppercase form. First, find all the files:

```bash
$ find . -name '*log'
.do-not-touch/important.log
admin.com-log/
foo.log
genius/spew.log
leather_flog
...
```

Oops, it looks like you need to include the dot in the pattern and to leave out directories as well. Do a <Control-P> to recall the command and then modify it:

```bash
$ find . -type f -name '*.*log'
.do-not-touch/important.log
foo.log
genius/spew.log
...
```

OK, this looks better. That .do-not-touch directory looks dangerous, though; you probably shouldn’t mess around in there:

```bash
$ find . -type f -name '*.*log' | grep -v .do-not-touch
foo.log
genius/spew.log
...
```

All right, that’s the exact list of files that need renaming. Try generating some new names:

```bash
$ find . -type f -name '*.*log' | grep -v .do-not-touch | while read fname
  > do
  >   echo mv $fname 'echo $fname | sed s/.log/.LOG/'
  > done
mv foo.log foo.LOG
mv genius/spew.log genius/spew.LOG
...
```
Yup, those are the commands to run to perform the renaming. So how to do it for real? You could recall the command and edit out the `echo`, which would make `sh` execute the `mv` commands instead of just printing them. However, piping the commands to a separate instance of `sh` is less error prone and requires less editing of the previous command.

When you do a `<Control-P>`, you’ll find that `bash` has thoughtfully collapsed your mini-script into a single line. To this condensed command line, simply add a pipe that sends the output to `sh -x`.

```
$ find . -type f -name '*.log' | grep -v .do-not-touch | while read fname;
    do echo mv $fname 'echo $fname | sed s/.log/.LOG/'; done | sh -x
+ mv foo.log foo.LOG
+ mv genius/spew.log genius/spew.LOG
...
```

The `-x` option to `sh` prints each command before executing it.

That completes the actual renaming, but save the script for future reuse. `bash`’s built-in command `fc` is a lot like `<Control-P>`, but instead of returning the last command to the command line, it transfers the command to your editor of choice. Add a shebang line and usage comment, write the file to a plausible location (`~/bin` or `/usr/local/bin`, perhaps), make the file executable, and you have a script.

To summarize this approach:

1. Develop the script (or script component) as a pipeline, one step at a time, entirely on the command line. Use `bash` for this process even though the eventual interpreter might be `dash` or another `sh` variant.

2. Send output to standard output and check to be sure it looks right.

3. At each step, use the shell’s command history to recall pipelines and the shell’s editing features to tweak them.

4. Until the output looks right, you haven’t actually done anything, so there’s nothing to undo if the command is incorrect.

5. Once the output is correct, execute the actual commands and verify that they worked as you intended.

6. Use `fc` to capture your work, then clean it up and save it.

In the example above, the command lines were printed and then piped to a subshell for execution. This technique isn’t universally applicable, but it’s often helpful. Alternatively, you can capture output by redirecting it to a file. No matter what, wait until you see the right stuff in the preview before doing anything that’s potentially destructive.
Input and output

The `echo` command is crude but easy. For more control over your output, use `printf`. It is a bit less convenient because you must explicitly put newlines where you want them (use “\n”), but it gives you the option to use tabs and enhanced number formatting in your the output. Compare the output from the following two commands:

```bash
$ echo "\taa\tbb\tcc\n"
\taa\tbb\tcc\n
$ printf "\taa\tbb\tcc\n"
   aa   bb   cc
```

Some systems have OS-level `printf` and `echo` commands, usually in `/usr/bin` and `/bin`, respectively. Although the commands and the shell built-ins are similar, they may diverge subtly in their specifics, especially in the case of `printf`. Either adhere to `sh`’s syntax or call the external `printf` with a full pathname.

You can use the `read` command to prompt for input. Here’s an example:

```bash
#!/bin/sh

echo -n "Enter your name: 
read user_name

if [ -n "$user_name" ]; then
    echo "Hello $user_name!"
    exit 0
else
    echo "Greetings, nameless one!"
    exit 1
fi
```

The `-n` in the `echo` command suppresses the usual newline, but you could also have used `printf` here. We cover the `if` statement’s syntax shortly, but its effect should be obvious here. The `-n` in the `if` statement evaluates to true if its string argument is not null. Here’s what the script looks like when run:

```bash
$ sh readexample
Enter your name: Ron
Hello Ron!
```
Spaces in filenames

The naming of files and directories is essentially unrestricted, except that names are limited in length and must not contain slash characters or nulls. In particular, spaces are permitted. Unfortunately, UNIX has a long tradition of separating command-line arguments at whitespace, so legacy software tends to break when spaces appear within filenames.

Spaces in filenames were once found primarily on filesystems shared with Macs and PCs, but they have now metastasized into UNIX culture and are found in some standard software packages as well. There are no two ways about it: administrative scripts must be prepared to deal with spaces in filenames (not to mention apostrophes, asterisks, and various other menacing punctuation marks).

In the shell and in scripts, spaceful filenames can be quoted to keep their pieces together. For example, the command

```
$ less "My spacey file"
```

preserves My spacey file as a single argument to less. You can also escape individual spaces with a backslash:

```
$ less My\ space\ file
```

The filename completion feature of most shells (usually bound to the <Tab> key) normally adds the backslashes for you.

When you are writing scripts, a useful weapon to know about is find’s -print0 option. In combination with xargs -0, this option makes the find/xargs combination work correctly regardless of the whitespace contained within filenames. For example, the command

```
$ find /home -type f -size +1M -print0 | xargs -0 ls -l
```

prints a long ls listing of every file beneath /home that’s over one megabyte in size.
Command-line arguments and functions

Command-line arguments to a script become variables whose names are numbers. $1$ is the first command-line argument, $s_2$ is the second, and so on. $s_0$ is the name by which the script was invoked. That could be a strange construction such as `../bin/example.sh`, so it doesn’t necessarily have the same value each time the script is run.

The variable $\#$ contains the number of command-line arguments that were supplied, and the variable $*$ contains all the arguments at once. Neither of these variables includes or counts $s_0$. Here’s an example of the use of arguments:

```bash
#!/bin/sh

show_usage() {
    echo "Usage: $0 source_dir dest_dir" 1>&2
    exit 1
}

# Main program starts here

if [ $# -ne 2 ]; then
    show_usage
else # There are two arguments
    if [ -d $1 ]; then
        source_dir=$1
    else
        echo 'Invalid source directory' 1>&2
        show_usage
    fi
    if [ -d $2 ]; then
        dest_dir=$2
    else
        echo 'Invalid destination directory' 1>&2
        show_usage
    fi
fi

printf "Source directory is ${source_dir}"
printf "Destination directory is ${dest_dir}"
```

If you call a script without arguments or with inappropriate arguments, the script should print a short usage message to remind you how to use it. The example script above accepts two arguments, validates that the arguments are both directories, and prints their names. If the arguments are invalid, the script prints a usage message and exits with a nonzero return code. If the caller of the script checks the return code, it will know that this script failed to execute correctly.

We created a separate `show_usage` function to print the usage message. If the script were later updated to accept additional arguments, the usage message would have to be changed in only one
place. The `1>&2` notation on lines that print error messages makes the output go to STDERR.

```bash
$ mkdir aaa bbb
$ sh showusage aaa bbb
Source directory is aaa
Destination directory is bbb
$ sh showusage foo bar
Invalid source directory
Usage: showusage source_dir dest_dir
```

Arguments to `sh` functions are treated like command-line arguments. The first argument becomes `$1`, and so on. As you can see above, `$0` remains the name of the script.

To make the example more robust, we could make the `show_usage` routine accept an error code as an argument. That would allow a more definitive code to be returned for each different type of failure. The next code excerpt shows how that might look.

```bash
define show_usage() {
    echo "Usage: $0 source_dir dest_dir" 1>&2
    if [ $# -eq 0 ]; then
        exit 99 # Exit with arbitrary nonzero return code
    else
        exit $1
    fi
}
```

In this version of the routine, the argument is optional. Within a function, `$#` tells you how many arguments were passed in. The script exits with code 99 if no more-specific code is designated. But a specific value, for example,

```bash
show_usage 5
```

makes the script exit with that code after printing the usage message. (The shell variable `$?` contains the exit status of the last command executed, whether used inside a script or at the command line.)

The analogy between functions and commands is strong in `sh`. You can define useful functions in your `~/.bash_profile` file (`~/.profile` for vanilla `sh`) and then use them on the command line as if they were commands. For example, if your site has standardized on network port 7988 for the SSH protocol (a form of “security through obscurity”), you might define

```bash
ssh() {
    /usr/bin/ssh -p 7988 $*
}
```

in your `~/.bash_profile` to make sure `ssh` is always run with the option `-p 7988`. 
Like many shells, **bash** has an aliasing mechanism that can reproduce this limited example even more concisely, but functions are more general and more powerful.
Control flow

We’ve seen several if-then and if-then-else forms in this chapter already; they do pretty much what you’d expect. The terminator for an if statement is fi. To chain your if clauses, you can use the elif keyword to mean “else if.” For example:

```plaintext
if [ $base -eq 1 ] && [ $dm -eq 1 ]; then
  installDMBase
elif [ $base -ne 1 ] && [ $dm -eq 1 ]; then
  installBase
elif [ $base -eq 1 ] && [ $dm -ne 1 ]; then
  installDM
else
  echo '==> Installing nothing'
fi
```

Both the peculiar [] syntax for comparisons and the command-line-optionlike names of the integer comparison operators (e.g., -eq) are inherited from the original Bourne shell’s channeling of `/bin/test`. The brackets are actually a shorthand way of invoking `test` and are not a syntactic requirement of the if statement. (In reality, these operations are now built into the shell and do not actually run `/bin/test`.)

Table 7.3 shows the sh comparison operators for numbers and strings. sh uses textual operators for numbers and symbolic operators for strings.

**Table 7.3: Elementary sh comparison operators**

<table>
<thead>
<tr>
<th>String</th>
<th>Numeric</th>
<th>True if</th>
</tr>
</thead>
<tbody>
<tr>
<td>$x = y</td>
<td>$x -eq y</td>
<td>$x is equal to $y</td>
</tr>
<tr>
<td>$x != y</td>
<td>$x -ne y</td>
<td>$x is not equal to $y</td>
</tr>
<tr>
<td>$x &lt; y</td>
<td>$x -lt y</td>
<td>$x is less than $y</td>
</tr>
<tr>
<td>n/a</td>
<td>$x -le y</td>
<td>$x is less than or equal to $y</td>
</tr>
<tr>
<td>$x &gt; y</td>
<td>$x -gt y</td>
<td>$x is greater than $y</td>
</tr>
<tr>
<td>n/a</td>
<td>$x -ge y</td>
<td>$x is greater than or equal to $y</td>
</tr>
<tr>
<td>-n $x</td>
<td>n/a</td>
<td>$x is not null</td>
</tr>
<tr>
<td>-z $x</td>
<td>n/a</td>
<td>$x is null</td>
</tr>
</tbody>
</table>

a. Must be backslash-escaped or double bracketed to prevent interpretation as an input or output redirection character.

sh shines in its options for evaluating the properties of files (once again, courtesy of its `/bin/test` legacy). Table 7.4 shows a few of sh’s many file testing and file comparison operators.

**Table 7.4: sh file evaluation operators**
Although the `elif` form is useful, a `case` selection is often a better choice for clarity. Its syntax is shown below in a sample routine that centralizes logging for a script. Of particular note are the closing parenthesis after each condition and the two semicolons that follow the statement block to be executed when a condition is met (except for the last condition). The `case` statement ends with ` esac`.

```
# The log level is set in the global variable LOG_LEVEL. The choices
# are, from most to least severe, Error, Warning, Info, and Debug.

logMsg() {
    message_level=$1
    message_itself=$2
    if [ $message_level -le $LOG_LEVEL ]; then
        case $message_level in
            0) message_level_text="Error" ;;
            1) message_level_text="Warning" ;;
            2) message_level_text="Info" ;;
            3) message_level_text="Debug" ;;
            *) message_level_text="Other" ;;
        esac
        echo "${message_level_text}: $message_itself"
    fi
}
```

This routine illustrates the common “log level” paradigm used by many administrative applications. The code of the script generates messages at many different levels of detail, but only the ones that pass a globally set threshold, `LOG_LEVEL`, are actually logged or acted on. To clarify the importance of each message, the message text is preceded by a label that denotes its associated log level.
Loops

sh’s for…in construct makes it easy to take some action for a group of values or files, especially when combined with filename globbing (the expansion of simple pattern-matching characters such as * and ? to form filenames or lists of filenames). The *.sh pattern in the for loop below returns a list of matching filenames in the current directory. The for statement then iterates through that list, assigning each filename in turn to the variable script.

```bash
#!/bin/sh

suffix=BACKUP--'date +%Y-%m-%d-%H%M'

for script in *.sh; do
    newname="$script.$suffix"
    echo "Copying $script to $newname..."
    cp -p $script $newname
done
```

The output looks like this:

```
$ sh forexample
Copying rhel.sh to rhel.sh.BACKUP--2017-01-28-2228...
Copying sles.sh to sles.sh.BACKUP--2017-01-28-2228...
...
```

The filename expansion is not magic in this context; it works exactly as it does on the command line. Which is to say, the expansion happens first and the line is then processed by the interpreter in its expanded form. (More accurately, the filename expansion is a little bit magical in that it does maintain a notion of the atomicity of each filename. Filenames that contain spaces go through the for loop in a single pass.) You could just as well have entered the filenames statically, as in the line

```bash
for script in rhel.sh sles.sh; do
```

In fact, any whitespace-separated list of things, including the contents of a variable, works as a target of for...in. You can also omit the list entirely (along with the in keyword), in which case the loop implicitly iterates over the script’s command-line arguments (if at the top level) or the arguments passed to a function:

```bash
#!/bin/sh

for file; do
    newname="${file}.backup"
    echo "Copying $file to $newname..."
    cp -p $file $newname
done
```
bash, but not vanilla sh, also has the more familiar for loop from traditional programming languages in which you specify starting, increment, and termination clauses.

For example:

```bash
# bash-specific
for (( i=0 ; i < $CPU_COUNT ; i++ )); do
    CPU_LIST="$CPU_LIST $i"
done
```

The next example illustrates sh’s while loop, which is useful for processing command-line arguments and for reading the lines of a file.

```bash
#!/bin/sh
exec 0<$1
counter=1
while read line; do
    echo "$counter: $line"
    counter=$((counter + 1))
done
```

Here’s what the output looks like:

```
$ sh whileexample /etc/passwd
1: root:x:0:0:Superuser:/root:/bin/bash
2: bin:x:1:1:bin:/bin:/bin/bash
3: daemon:x:2:2:Daemon:/sbin:/bin/bash
...
```

This scriptlet has a couple of interesting features. The exec statement redefines the script’s standard input to come from whatever file is named by the first command-line argument. The file must exist or the script generates an error. Depending on the invocation, exec can also have the more familiar meaning “stop this script and transfer control to another script or expression.” It’s yet another shell oddity that both functions are accessed through the same statement.

The read statement within the while clause is a shell built-in, but it acts like an external command. You can put external commands in a while clause as well; in that form, the while loop terminates when the external command returns a nonzero exit status.

The $((counter + 1)) expression is an odd duck, indeed. The $((...)) notation forces numeric evaluation. It also makes optional the use of $ to mark variable names. The expression is replaced with the result of the arithmetic calculation.

The $((...)) shenanigans work in the context of double quotes, too. In bash, which supports C’s ++ postincrement operator, the body of the loop can be collapsed down to one line.
while read line; do
    echo "${counter++}: $line"
done
Arithmetic

All `sh` variables are string valued, so `sh` does not distinguish between the number 1 and the character string “1” in assignments. The difference lies in how the variables are used. The following code illustrates the distinction:

```bash
#!/bin/sh
a=1
b=$((2))
c=$a+$b
d=$((a + b))

echo "$a + $b = $c \t(plus sign as string literal)"
echo "$a + $b = $d \t(plus sign as arithmetic addition)"
```

This script produces the output

```
1 + 2 = 1+2  (plus sign as string literal)
1 + 2 = 3    (plus sign as arithmetic addition)
```

Note that the plus sign in the assignment to $c does not act as a concatenation operator for strings. It’s just a literal character. That line is equivalent to

```
c="$a+$b"
```

To force numeric evaluation, you enclose an expression in `$((...))`, as shown with the assignment to $d above. But even this precaution does not result in $d receiving a numeric value; the result of the calculation is the string “3”.

`sh` has the usual assortment of arithmetic, logical, and relational operators; see the man page for details.
7.4 Regular Expressions

As we mentioned here, regular expressions are standardized patterns that parse and manipulate text. For example, the regular expression

\texttt{I sent you a che(que|ck) for the gr[ae]y-colou?red alumini?um.}

matches sentences that use either American or British spelling conventions.

Regular expressions are supported by most modern languages, though some take them more to heart than others. They’re also used by UNIX commands such as \texttt{grep} and \texttt{vi}. They are so common that the name is usually shortened to “regex.” Entire books have been written about how to harness their power; you can find citations for two of them at the end of this chapter.

The filename matching and expansion performed by the shell when it interprets command lines such as \texttt{wc -l *.pl} is not a form of regular expression matching. It’s a different system called “shell globbing,” and it uses a different and simpler syntax.

Regular expressions are not themselves a scripting language, but they’re so useful that they merit featured coverage in any discussion of scripting; hence, this section.
The matching process

Code that evaluates a regular expression attempts to match a single given text string to a single given pattern. The “text string” to match can be very long and can contain embedded newlines. It’s sometimes convenient to use a regex to match the contents of an entire file or document.

For the matcher to declare success, the entire search pattern must match a contiguous section of the search text. However, the pattern can match at any position. After a successful match, the evaluator returns the text of the match along with a list of matches for any specially delimited subsections of the pattern.
**Literal characters**

In general, characters in a regular expression match themselves. So the pattern

```
I am the walrus
```

matches the string “I am the walrus” and that string only. Since it can match anywhere in the search text, the pattern can be successfully matched to the string

```
I am the egg man. I am the walrus. Koo koo ka-choo!
```

However, the actual match is limited to the “I am the walrus” portion. Matching is case sensitive.
Special characters

Table 7.5 shows the meanings of some common special symbols that can appear in regular expressions. These are just the basics—there are many more.

Table 7.5: Special characters in regular expressions (common ones)

<table>
<thead>
<tr>
<th>Symbol</th>
<th>What it matches or does</th>
</tr>
</thead>
<tbody>
<tr>
<td>.</td>
<td>Matches any character</td>
</tr>
<tr>
<td>[chars]</td>
<td>Matches any character from a given set</td>
</tr>
<tr>
<td>[^chars]</td>
<td>Matches any character not in a given set</td>
</tr>
<tr>
<td>\w</td>
<td>Matches any “word” character (same as [A-Za-z0-9_])</td>
</tr>
<tr>
<td>\s</td>
<td>Matches any whitespace character (same as [ \t\n\r]) a</td>
</tr>
<tr>
<td>\d</td>
<td>Matches any digit (same as [0-9])</td>
</tr>
<tr>
<td></td>
<td></td>
</tr>
<tr>
<td>(expr)</td>
<td>Limits scope, groups elements, allows matches to be captured</td>
</tr>
<tr>
<td>?</td>
<td>Allows zero or one match of the preceding element</td>
</tr>
<tr>
<td>*</td>
<td>Allows zero, one, or many matches of the preceding element</td>
</tr>
<tr>
<td>+</td>
<td>Allows one or more matches of the preceding element</td>
</tr>
<tr>
<td>{ n }</td>
<td>Matches exactly n instances of the preceding element</td>
</tr>
<tr>
<td>{ min, }</td>
<td>Matches at least min instances (note the comma)</td>
</tr>
<tr>
<td>{ min,max }</td>
<td>Matches any number of instances from min to max</td>
</tr>
</tbody>
</table>

a. That is, a space, a form feed, a tab, a newline, or a return

Many special constructs, such as + and |, affect the matching of the “thing” to their left or right. In general, a “thing” is a single character, a subpattern enclosed in parentheses, or a character class enclosed in square brackets. For the | character, however, thingness extends indefinitely to both left and right. If you want to limit the scope of the vertical bar, enclose the bar and both things in their own set of parentheses. For example,

I am the (walrus|egg man)\.

matches either “I am the walrus.” or “I am the egg man.”. This example also demonstrates escaping of special characters (here, the dot). The pattern

(I am the (walrus\|egg man)\. \?)\{1,2\}

matches any of the following:

- I am the walrus.
- I am the egg man.
- I am the walrus. I am the egg man.
- I am the egg man. I am the walrus.
- I am the egg man. I am the egg man.
- I am the walrus. I am the walrus.

It also matches “I am the walrus. I am the egg man. I am the walrus.”, even though the number of repetitions is explicitly capped at two. That’s because the pattern need not match the entire search text. Here, the regex matches two sentences and terminates, declaring success. It doesn’t care that another repetition is available.

It is a common error to confuse the regular expression metacharacter * (the zero-or-more quantifier) with the shell’s * globbing character. The regex version of the star needs something to modify; otherwise, it won’t do what you expect. Use .* if any sequence of characters (including no characters at all) is an acceptable match.
Example regular expressions

In the United States, postal ("zip") codes have either five digits or five digits followed by a dash and four more digits. To match a regular zip code, you must match a five-digit number. The following regular expression fits the bill:

```
^\d\{5\}$
```

The ^ and $ match the beginning and end of the search text but do not actually correspond to characters in the text; they are “zero-width assertions.” These characters ensure that only texts consisting of exactly five digits match the regular expression—the regex will not match five digits within a larger string. The \d escape matches a digit, and the quantifier \{5\} says that there must be exactly five one-digit matches.

To accommodate either a five-digit zip code or an extended zip+4, add an optional dash and four additional digits:

```
^\d\{5\}(\-\d\{4\})?$
```

The parentheses group the dash and extra digits together so that they are considered one optional unit. For example, the regex won’t match a five-digit zip code followed by a dash. If the dash is present, the four-digit extension must be present as well or there is no match.

A classic demonstration of regex matching is the following expression,

```
M[ou]'?am+[ae]\r ([AEae]\[\-\])?[GKQ]h?[aeu]+([dtz]\[dhz\]?){1,2}af[iy]
```

which matches most of the variant spellings of the name of former Libyan head of state Moammar Gadhafi, including

- Muammar al-Kaddafı (BBC)
- Moammar Gadhafi (Associated Press)
- Muammar al-Qadhafi (Al-Jazeera)
- Mu’ammar Al-Qadhafi (U.S. Department of State)

Do you see how each of these would match the pattern? Note that this regular expression is designed to be liberal in what it matches. Many patterns that aren’t legitimate spellings also match: for example, “Mo’ammer el Qhuuuzzthaf”.

This regular expression also illustrates how quickly the limits of legibility can be reached. Most regex systems support an \x option that ignores literal whitespace in the pattern and enables comments, allowing the pattern to be spaced out and split over multiple lines. You can then use whitespace to separate logical groups and clarify relationships, just as you would in a procedural language. For example, here’s a more readable version of that same Moammar Gadhafi regex:
This helps a bit, but it’s still pretty easy to torture later readers of your code. So be kind: if you can, use hierarchical matching and multiple small matches instead of trying to cover every possible situation in one large regular expression.
Captures

When a match succeeds, every set of parentheses becomes a “capture group” that records the actual text that it matched. The exact manner in which these pieces are made available to you depends on the implementation and context. In most cases, you can access the results as a list, array, or sequence of numbered variables.

Since parentheses can nest, how do you know which match is which? Easy: the matches arrive in the same order as the opening parentheses. There are as many captures as there are opening parentheses, regardless of the role (or lack of role) that each parenthesized group played in the actual matching. When a parenthesized group is not used (e.g., Mu(')?ammar when matched against “Muammar”), its corresponding capture is empty.

If a group is matched more than once, the contents of only the last match are returned. For example, with the pattern

(I am the (walrus|egg man)\. \?)\{1,2\}

matching the text

I am the egg man. I am the walrus.

there are two results, one for each set of parentheses:

I am the walrus.

walrus

Both capture groups actually matched twice. However, only the last text to match each set of parentheses is actually captured.
Greediness, laziness, and catastrophic backtracking

Regular expressions match from left to right. Each component of the pattern matches the longest possible string before yielding to the next component, a characteristic known as greediness.

If the regex evaluator reaches a state from which a match cannot be completed, it unwinds a bit of the candidate match and makes one of the greedy atoms give up some of its text. For example, consider the regex \(a^*aa\) being matched against the input text “aaaaaa”.

At first, the regex evaluator assigns the entire input to the \(a^*\) portion of the regex because the \(a^*\) is greedy. When there are no more \(a\)’s to match, the evaluator goes on to try to match the next part of the regex. But oops, it’s an \(a\), and there is no more input text that can match an \(a\); time to backtrack. The \(a^*\) has to give up one of the \(a\)’s it has matched.

Now the evaluator can match \(a^*a\), but it still cannot match the last \(a\) in the pattern. So it backtracks again and takes away a second \(a\) from the \(a^*\). Now the second and third \(a\)’s in the pattern both have \(a\)’s to pair with, and the match is complete.

This simple example illustrates some important general points. First, greedy matching plus backtracking makes it expensive to match apparently simple patterns such as `<img.*></tr>` when processing entire files. The `.*` portion starts by matching everything from the first `<img` to the end of the input, and only through repeated backtracking does it contract to fit the local tags.

Furthermore, the `</tr>` that this pattern binds to is the last possible valid match in the input, which is probably not what you want. More likely, you meant to match an `<img>` tag followed immediately by a `</tr>` tag. A better way to write this pattern is `<img[^>]*?>\s*</tr>`, which allows the initial wild card match to expand only to the end of the current tag, because it cannot cross a right-angle-bracket boundary.

You can also use lazy (as opposed to greedy) wild card operators: `*?` instead of `*`, and `+?` instead of `+`. These versions match as few characters of the input as they can. If that fails, they match more. In many situations, these operators are more efficient and closer to what you want than the greedy versions.

Note, however, that they can produce matches different from those of the greedy operators; the difference is more than just one of implementation. In our HTML example, the lazy pattern would be `<img[^>]*?><tr>`. But even here, the `.*?` could eventually grow to include unwanted `>`’s because the next tag after an `<img` might not be a `</tr>`. Again, probably not what you want.

Patterns with multiple wild card sections can cause exponential behavior in the regex evaluator, especially if portions of the text can match several of the wild card expressions and especially if the search text does not match the pattern. This situation is not as unusual as it might sound, especially when pattern matching with HTML. Often, you’ll want to match certain tags followed by other tags, possibly separated by even more tags, a recipe that might require the regex evaluator to try many possible combinations.
Regex guru Jan Goyvaerts calls this phenomenon “catastrophic backtracking” and writes about it in his blog; see regular-expressions.info/catastrophic.html for details and some good solutions.

(Although this section shows HTML excerpts as examples of text to be matched, regular expressions are not really the right tool for this job. Our external reviewers were uniformly aghast. Ruby and Python both have excellent add-ons that parse HTML documents the proper way. You can then access the portions you’re interested in with XPath or CSS selectors. See the Wikipedia page for XPath and the respective languages’ module repositories for details.)

A couple of take-home points from all this:

- If you can do pattern matching line-by-line rather than file-at-a-time, there is much less risk of poor performance.
- Even though regex notation makes greedy operators the default, they probably shouldn’t be. Use lazy operators.
- All uses of .* are inherently suspicious and should be scrutinized.
7.5 **Python programming**

Python and Ruby are interpreted languages with a pronounced object-oriented inflection. Both are widely used as general-purpose scripting languages and have extensive libraries of third party modules. We discuss Ruby in more detail starting here.

Python offers a straightforward syntax that’s usually pretty easy to follow, even when reading other people’s code.

We recommend that all sysadmins become fluent in Python. It’s the modern era’s go-to language for both system administration and general-purpose scripting. It’s also widely supported as a glue language for use within other systems (e.g., the PostgreSQL database and Apple’s Xcode development environment). It interfaces cleanly with REST APIs and has well-developed libraries for machine learning, data analysis, and numeric computation.
The passion of Python 3

Python was already well on its way to becoming the world’s default scripting language when Python 3 was released in 2008. For this release, the developers chose to forgo backward compatibility with Python 2 so that a group of modest but fundamental changes and corrections could be made to the language, particularly in the area of internationalized text processing. The exact list of changes in Python 3 isn’t relevant to this brief discussion, but you can find a summary at docs.python.org/3.0/whatsnew/3.0.html.

Unfortunately, the rollout of Python 3 proved to be something of a debacle. The language updates are entirely sensible, but they’re not must-haves for the average Python programmer with an existing codebase to maintain. For a long time, scripters avoided Python 3 because their favorite libraries didn’t support it, and library authors didn’t support Python 3 because their clients were still using Python 2.

Even in the best of circumstances, it’s difficult to push a large and interdependent user community past this sort of discontinuity. In the case of Python 3, early entrenchments persisted for the better part of a decade. However, as of 2017, that situation finally seems to be changing.

Compatibility libraries that allow the same Python code to run under either version of the language have helped ease the transition, to some extent. But even now, Python 3 remains less common in the wild than Python 2.

As of this writing, py3readiness.org reports that only 17 of the top 360 Python libraries remain incompatible with Python 3. But the long tail of unported software is more sobering: only a tad more than 25% of the libraries warehoused at pypi.python.org (the Python Package Index, aka PyPI) run under Python 3. Of course, many of these projects are older and no longer maintained, but 25% is still a concerningly low number. See caniusepython3.com for up-to-date statistics.
Python 2 or Python 3?

The world’s solution to the slowly unfolding Python transition has been to treat Pythons 2 and 3 as separate languages. You needn’t consecrate your systems to one or the other; you can run both simultaneously without conflict.

All our example systems ship Python 2 by default, usually as /usr/bin/python2 with a symbolic link from /usr/bin/python. Python 3 can typically be installed as a separate package; the binary is called python3.

Although the Fedora project is working to make Python 3 its system default, Red Hat and CentOS are far behind and do not even define a prebuilt package for Python 3. However, you can pick one up from Fedora’s EPEL (Extra Packages for Enterprise Linux) repository. See the FAQ at fedoraproject.org/wiki/EPEL for instructions on accessing this repository. It’s easy to set up, but the exact commands are version-dependent.

For new scripting work or for those new to Python altogether, it makes sense to jump directly to Python 3. That’s the syntax we show in this chapter, though in fact it’s only the print lines that vary between Python 2 and Python 3 in our simple examples.

For existing software, use whichever version of Python the software prefers. If your choice is more complicated than simply new vs. old code, consult the Python wiki at wiki.python.org/moin/Python2orPython3 for an excellent collection of issues, solutions, and recommendations.
**Python quick start**

For a more thorough introduction to Python than we can give here, Mark Pilgrim’s *Dive Into Python 3* is a great place to start. It’s available for reading or for download (without charge) at diveintopython3.net, or as a printed book from Apress. A complete citation can be found [here](#).

To start, here’s a quick “Hello, world!” script:

```bash
#!/usr/bin/python3
print("Hello, world!")
```

To get it running, set the execute bit or invoke the `python3` interpreter directly:

```
$ chmod +x helloworld
$ ./helloworld
Hello, world!
```

Python’s most notable break with tradition is that indentation is logically significant. Python does not use braces, brackets, or `begin` and `end` to delineate blocks. Statements at the same level of indentation automatically form blocks. The exact indentation style (spaces or tabs, depth of indentation) does not matter.

Python blocking is best shown by example. Consider this simple if-then-else statement:

```python
import sys
a = sys.argv[1]
if a == "1":
    print('a is one')
    print('This is still the then clause of the if statement.')
else:
    print('a is', a)
    print('This is still the else clause of the if statement.')
print('This is after the if statement.')
```

The first line imports the `sys` module, which contains the `argv` array. The two paths through the `if` statement both have two lines, each indented to the same level. (Colons at the end of a line are normally a clue that the line introduces and is associated with an indented block that follows it.) The final `print` statement lies outside the context of the `if` statement.
Python’s indentation convention is less flexible for the formatting of code, but it does reduce clutter in the form of braces and semicolons. It’s an adjustment for those accustomed to traditional delimiters, but most people ultimately find that they like it.

Python’s `print` function accepts an arbitrary number of arguments. It inserts a space between each pair of arguments and automatically supplies a newline. You can suppress or modify these characters by adding `end=` or `sep=` options to the end of the argument list.

For example, the line

```
print("one", "two", "three", sep="-", end="!\n")
```

produces the output

```
one-two-three!
```

Comments are introduced with a sharp (#) and last until the end of the line, just as in `sh`, Perl, and Ruby.

You can split long lines by backslashing the end of line breaks. When you do this, the indentation of only the first line is significant. You can indent the continuation lines however you like. Lines with unbalanced parentheses, square brackets, or curly braces automatically signal continuation even in the absence of backslashes, but you can include the backslashes if doing so clarifies the structure of the code.

Some cut-and-paste operations convert tabs to spaces, and unless you know what you’re looking for, this can drive you nuts. The golden rule is never to mix tabs and spaces; use one or the other for indentation. A lot of software makes the traditional assumption that tabs fall at 8-space intervals, which is too much indentation for readable code. Most in the Python community seem to prefer spaces and 4-character indentation.

However you decide to attack the indentation problem, most editors have options that can help save your sanity, either by outlawing tabs in favor of spaces or by displaying spaces and tabs differently. As a last resort, you can translate tabs to spaces with the `expand` command.
Objects, strings, numbers, lists, dictionaries, tuples, and files

All data types in Python are objects, and this gives them more power and flexibility than they have in most languages.

In Python, lists are enclosed in square brackets and indexed from zero. They are essentially similar to arrays, but can hold objects of any type. (A homogeneous and more efficient array type is implemented in the array module, but for most purposes, stick with lists.)

Python also has “tuples,” which are essentially immutable lists. Tuples are faster than lists and are helpful for representing constant data. The syntax for tuples is the same as for lists, except that the delimiters are parentheses instead of square brackets. Because (thing) looks like a simple algebraic expression, tuples that contain only a single element need a marker comma to disambiguate them: (thing,).

Here’s some basic variable and data type wrangling in Python:

```python
name = 'Gwen'
rating = 10
characters = [ 'SpongeBob', 'Patrick', 'Squidward' ]
elements = ( 'lithium', 'carbon', 'boron' )

print("name:\t%s\nrating:\t%d" % (name, rating))
print("characters:\t%s" % characters)
print("hero:\t%s" % characters[0])
print("elements:\t%s" % (elements,))
```

This example produces the following output:

```
$ python3 objects
name:     Gwen
rating:   10
characters:   ['SpongeBob', 'Patrick', 'Squidward']
hero:     SpongeBob
elements: ('lithium', 'carbon', 'boron')
```

Note that the default string conversion for list and tuple types represents them as they would be found in source code.

Variables in Python are not syntactically marked or declared by type, but the objects to which they refer do have an underlying type. In most cases, Python does not automatically convert types for you, but individual functions or operators may do so. For example, you cannot concatenate a string and a number (with the + operator) without explicitly converting the number to its string representation. However, formatting operators and statements coerce everything to string form.
Every object has a string representation, as can be seen in the output above. Dictionaries, lists, and tuples compose their string representations recursively by stringifying their constituent elements and combining these strings with the appropriate punctuation.

The string formatting operator % is a lot like the sprintf function from C, but it can be used anywhere a string can appear. It’s a binary operator that takes the string on its left and the values to be inserted on its right. If more than one value is to be inserted, the values must be presented as a tuple.

A Python dictionary (also known as a hash or an associative array) represents a set of key/value pairs. You can think of a hash as an array whose subscripts (keys) are arbitrary values; they do not have to be numbers. But in practice, numbers and strings are common keys.

Dictionary literals are enclosed in curly braces, with each key/value pair being separated by a colon. In use, dictionaries work much like lists, except that the subscripts (keys) can be objects other than integers.

```python
ordinal = { 1 : 'first', 2 : 'second', 3 : 'third' }
print("The ordinal dictionary contains", ordinal)
print("The ordinal of 1 is", ordinal[1])
```

```
$ python3 dictionary
The ordinal array contains {1: 'first', 2: 'second', 3: 'third'}
The ordinal of 1 is first
```

Python handles open files as objects with associated methods. True to its name, the readline method reads a single line, so the example below reads and prints two lines from the /etc/passwd file.

```python
f = open('/etc/passwd', 'r')
print(f.readline(), end="\n")
print(f.readline(), end="\n")
f.close()
```

```
$ python3 fileio
root:x:0:0:root:/root:/bin/bash
daemon:x:1:1:daemon:/usr/sbin:/usr/sbin/nologin
```

The newlines at the end of the print calls are suppressed with end="\n" because each line already includes a newline character from the original file. Python does not automatically strip these.
**Input validation example**

Our scriptlet below shows a general scheme for input validation in Python. It also demonstrates the definition of functions and the use of command-line arguments, along with a couple of other Pythonisms.

```python
import sys
import os

def show_usage(message, code = 1):
    print(message)
    print("%s: source_dir dest_dir" % sys.argv[0])
    sys.exit(code)

if len(sys.argv) != 3:
    show_usage("2 args required; you supplied %d" % (len(sys.argv) - 1))
elif not os.path.isdir(sys.argv[1]):
    show_usage("Invalid source directory")
elif not os.path.isdir(sys.argv[2]):
    show_usage("Invalid destination directory")

source, dest = sys.argv[1:3]

print("Source directory is", source)
print("Destination directory is", dest)
```

In addition to importing the `sys` module, we also import the `os` module to gain access to the `os.path.isdir` routine. Note that `import` doesn’t shortcut your access to any symbols defined by modules; you must use fully qualified names that start with the module name.

The definition of the `show_usage` routine supplies a default value for the exit code in case the caller does not specify this argument explicitly. Since all data types are objects, function arguments are effectively passed by reference.

The `sys.argv` list contains the script name in the first position, so its length is one greater than the number of command-line arguments that were actually supplied. The form `sys.argv[1:3]` is a list slice. Curiously, slices do not include the element at the far end of the specified range, so this slice includes only `sys.argv[1]` and `sys.argv[2]`. You could simply say `sys.argv[1:]` to include the second and subsequent arguments.

Like `sh`, Python has a dedicated “else if” condition; the keyword is `elif`. There is no explicit case or switch statement.

The parallel assignment of the `source` and `dest` variables is a bit different from some languages in that the variables themselves are not in a list. Python allows parallel assignments in either form.

Python uses the same comparison operators for numeric and string values. The “not equal” comparison operator is `!=`, but there is no unary `!` operator; use `not` for this. The Boolean operators
and and or are also spelled out.
Loops

The fragment below uses a `for...in` construct to iterate through the range 1 to 10.

```python
for counter in range(1, 10):
    print(counter, end=" ")
print()                     # Add final newline
```

As with the array slice in the previous example, the right endpoint of the range is not actually included. The output includes only the numbers 1 through 9:

1 2 3 4 5 6 7 8 9

This is Python’s only type of for loop, but it’s a powerhouse. Python’s `for` has several features that distinguish it from `for` in other languages:

- Nothing is special about numeric ranges. Any object can support Python’s iteration model, and most common objects do. You can iterate through a string (by character), a list, a file (by character, line, or block), a list slice, etc.

- Iterators can yield multiple values, and you can have multiple loop variables. The assignment at the top of each iteration acts just like Python’s regular multiple assignments. This feature is particularly nice for iterating through dictionaries.

- Both `for` and `while` loops can have `else` clauses at the end. The `else` clause is executed only if the loop terminates normally, as opposed to exiting through a `break` statement. This feature may initially seem counterintuitive, but it handles certain use cases quite elegantly.

The example script below accepts a regular expression on the command line and matches it against a list of Snow White’s dwarves and the colors of their dwarf suits. The first match is printed, with the portions that match the regex surrounded by underscores.
import sys
import re
suits = {
    'Bashful':'yellow', 'Sneezy':'brown', 'Doc':'orange', 'Grumpy':'red',
    'Dopey':'green', 'Happy':'blue', 'Sleepy':'taupe'
}
pattern = re.compile("\(%s\)" % sys.argv[1])
for dwarf, color in suits.items():
    if pattern.search(dwarf) or pattern.search(color):
        print("%s's dwarf suit is %s.\n" %
            (pattern.sub(r"_\1_", dwarf), pattern.sub(r"_\1_", color)))
        break
else:
    print("No dwarves or dwarf suits matched the pattern.")

Here’s some sample output:

$ python3 dwarfsearch '^[aeiou]{2}$'
Sl_ee_py's dwarf suit is t_au_pe.
$ python3 dwarfsearch '^[ga]gu$'
No dwarves or dwarf suits matched the pattern.

The assignment to suits demonstrates Python’s syntax for encoding literal dictionaries. The
suits.items() method is an iterator for key/value pairs—note that we’re extracting both a dwarf name
and a suit color on each iteration. If you wanted to iterate through only the keys, you could just
say for dwarf in suits.

Python implements regular expression handling through its re module. No regex features are built
into the language itself, so regex-wrangling with Python is a bit clunkier than with, say, Perl.
Here, the regex pattern is initially compiled from the first command-line argument surrounded by
parentheses (to form a capture group). Strings are then tested and modified with the search and sub
methods of the regex object. You can also call re.search et al. directly as functions, supplying the
regex to use as the first argument.

The \1 in the substitution string is a back-reference to the contents of the first capture group. The
strange-looking r prefix that precedes the substitution string (r"_\1_") suppresses the normal
substitution of escape sequences in string constants (r stands for “raw”). Without this, the
replacement pattern would consist of two underscores surrounding a character with numeric code
1.

One thing to note about dictionaries is that they have no defined iteration order. If you run the
dwarf search a second time, you may well receive a different answer:

$ python3 dwarfsearch '^[aeiou]{2}$'
Dopey's dwarf suit is gr_ee_n.
7.6 Ruby Programming

Ruby, designed and maintained by Japanese developer Yukihiro “Matz” Matsumoto, shares many features with Python, including a pervasive “everything’s an object” approach. Although initially released in the mid-1990s, Ruby did not gain prominence until a decade later with the release of the Rails web development platform.

Ruby is still closely associated with the web in many people’s minds, but there’s nothing web-specific about the language itself. It works well for general-purpose scripting. However, Python is probably a better choice for a primary scripting language, if only because of its wider popularity.

Although Ruby is roughly equivalent to Python in many ways, it is philosophically more permissive. Ruby classes remain open for modification by other software, for example, and the Rubyist community attaches little or no shame to extensions that modify the standard library.

Ruby appeals to those with a taste for syntactic sugar, features that don’t really change the basic language but that permit code to be expressed more concisely and clearly. In the Rails environment, for example, the line

due_date = 7.days.from_now

creates a Time object without referencing the names of any time-related classes or doing any explicit date-and-time arithmetic. Rails defines days as an extension to Fixnum, the Ruby class that represents integers. This method returns a Duration object that acts like a number; used as a value, it’s equivalent to 604,800, the number of seconds in seven days. Inspected in the debugger, it describes itself as “7 days.” (This form of polymorphism is common to both Ruby and Python. It’s often called “duck typing”; if an object walks like a duck and quacks like a duck, you needn’t worry about whether it’s actually a duck.)

See Chapter 23 for more information about Chef and Puppet.

Ruby makes it easy for developers to create “domain-specific languages” (aka DSLs), mini-languages that are in fact Ruby but that read like specialized configuration systems. Ruby DSLs are used to configure both Chef and Puppet, for example.
Installation

Some systems have Ruby installed by default and some do not. However, it’s always available as a package, often in several versions.

To date (version 2.3), Ruby has maintained relatively good compatibility with old code. In the absence of specific warnings, it’s generally best to install the most recent version.

See this page for more about RVM.

Unfortunately, most systems’ packages lag several releases behind the Ruby trunk. If your package library doesn’t include the current release (check ruby-lang.org to determine what that is), install the freshest version through RVM; don’t try to do it yourself.
Ruby quick start

Since Ruby is so similar to Python, here a perhaps-eerily-familiar look at some Ruby snippets modeled on those from the Python section earlier in this chapter.

```ruby
#!/usr/bin/env ruby

print "Hello, world!\n\n"

name = 'Gwen'
rating = 10
characters = [ 'SpongeBob', 'Patrick', 'Squidward' ]
elements = { 3 => 'lithium', 7 => 'carbon', 5 => 'boron' }

print "Name:\t", name, "\nRating:\t", rating, "\n"
print "Characters:\t#{characters}\n"
print "Elements:\t#{elements}\n"

element_names = elements.values.sort!.map(&:upcase).join(', ')
print "Element names:\t", element_names, "\n"

elements.each do |key, value|
  print "Atomic number #{key} is #{value}.\n"
end
```

The output is as follows:

Hello, world!

Name: Gwen
Rating: 10
Characters: ["SpongeBob", "Patrick", "Squidward"]
Elements: {3=>"lithium", 7=>"carbon", 5=>"boron"}

Element names: BORON, CARBON, LITHIUM

Atomic number 3 is lithium.
Atomic number 7 is carbon.
Atomic number 5 is boron.

Like Python, Ruby uses brackets to delimit arrays and curly braces to delimit dictionary literals. (Ruby calls them “hashes.”) The => operator separates each hash key from its corresponding value, and the key/value pairs are separated from each other by commas. Ruby does not have tuples.

Ruby’s print is a function (or more accurately, a global method), just like that of Python 3. However, if you want newlines, you must specify them explicitly. There’s also a puts function that adds newlines for you, but it’s perhaps a bit too smart. If you try to add an extra newline of your own, puts won’t insert its own newline.
The parentheses normally seen around the arguments of function calls are optional in Ruby. Developers don’t normally include them unless they help to clarify or disambiguate the code. (Note that some of these calls to `print` do include multiple arguments separated by commas.)

In several cases, we’ve used `#{}` brackets to interpolate the values of variables into double-quoted strings. Such brackets can contain arbitrary Ruby code; whatever value the code produces is automatically converted to string type and inserted into the outer string. You can also concatenate strings with the `+` operator, but interpolation is typically more efficient.

The line that calculates `element_names` illustrates several more Ruby tropes:

```
element_names = elements.values.sort!.map(&:upcase).join(' ', '')
```

This is a series of method calls, each of which operates on the result returned by the previous method, much like a series of pipes in the shell. For example, `elements`’ `values` method produces an array of strings, which `sort!` then orders alphabetically. This array’s `map` method calls the `upcase` method on each element, then reassembles all the results back into a new array. Finally, `join` concatenates the elements of that array, interspersed with commas, to produce a string.

The bang at the end of `sort!` warns you that there’s something to be wary of when using this method. It isn’t significant to Ruby; it’s just part of the method’s name. In this case, the issue of note is that `sort!` sorts the array in place. There’s also a `sort` method (without the `!`) that returns the elements in a new, sorted array.
Blocks

In the previous code example, the text between do and end is a block, also commonly known in other languages as a lambda function, a closure, or an anonymous function:

```ruby
elements.each do |key, value|
    print "Atomic number #{key} is #{value}.\n"
end
```

Ruby actually has three entities of this general type, known as blocks, procs, and lambdas. The differences among them are subtle and not important for this overview. The block above takes two arguments, which it calls key and value. It prints the values of both.

`each` looks like it might be a language feature, but it’s just a method defined by hashes. `each` accepts the block as an argument and calls it once for each key/value pair the hash contains. This type of iteration function used in combination with a block is highly characteristic of Ruby code. `each` is the standard name for generic iterators, but many classes define more specific versions such as `each_line` or `each_character`.

Ruby has a second syntax for blocks that uses curly braces instead of `do...end` as delimiters. It means exactly the same thing, but it looks more at home as part of an expression. For example,

```ruby
characters.map { |c| c.reverse } # ["boBegnopS", "kcirtaP", "drowdiuqS"]
```

This form is functionally identical to `characters.map(&:reverse)`, but instead of just telling `map` what method to call, we included an explicit block that calls the `reverse` method.

The value of a block is the value of the last expression it evaluates before completing. Conveniently, pretty much everything in Ruby is an expression (meaning “a piece of code that can be evaluated to produce a value”), including control structures such as `case` (analogous to what most languages call `switch`) and `if-else`. The values of these expressions mirror the value produced by whichever case or branch was activated.

Blocks have many uses other than iteration. They let a single function perform both setup and takedown procedures on behalf of another section of code, so they often represent multi-step operations such as database transactions or filesystem operations.

For example, the following code opens the `/etc/passwd` file and prints out the line that defines the root account:

```ruby
open '/etc/passwd', 'r' do |file|
    file.each_line do |line|
        print line if line.start_with? 'root:'
    end
end
```
The `open` function opens the file and passes its IO object to the outer block. Once the block has finished running, `open` automatically closes the file. There’s no need for a separate `close` operation (although it does exist if you want to use it), and the file is closed no matter how the outer block terminates.

The postfix `if` construct used here might be familiar to those who have used Perl. It’s a nice way to express simple conditionals without obscuring the primary action. Here, it’s clear at a glance that the inner block is a loop that prints out some of the lines.

In case the structure of that `print` line is not clear, here it is again with the optional parentheses included. The `if` has the lowest precedence, and it has a single method call on either side:

```ruby
   print(line) if line.start_with?('root:')
```

As with the `sort!` method we saw on this page, the question mark is just a naming convention for methods that return Boolean values.

The syntax for defining a named function is slightly different from that for a block:

```ruby
def show_usage(msg = nil)
   STDERR.puts msg if msg
   STDERR.puts "Usage: #$0 filename ..."
   exit 1
end
```

The parentheses are still optional, but in practice, they are always shown in this context unless the function takes no arguments. Here, the `msg` argument defaults to `nil`.

The global variable `$0` is magic and contains the name by which the current program was invoked. (Traditionally, this would be the first argument of the `argv` array. But the Ruby convention is that `ARGV` contains only actual command-line arguments.)

As in C, you can treat non-Boolean values as if they were Booleans, as illustrated here in the form of `if msg`. The Ruby mapping for this conversion is a bit unusual, though: everything except `nil` and `false` counts as true. In particular, `0` is true. (In practice, this usually ends up being what you want.)
Symbols and option hashes

Ruby makes extensive use of an uncommon data type called a symbol, denoted with a colon, e.g., :example. You can think of symbols as immutable strings. They’re commonly used as labels or as well-known hash keys. Internally, Ruby implements them as numbers, so they’re fast to hash and compare.

Symbols are so commonly used as hash keys that Ruby 2.0 defined an alternative syntax for hash literals to reduce the amount of punctuation clutter. The standard-form hash

```ruby
h = { :animal => 'cat', :vegetable => 'carrot', :mineral => 'zeolite' }
```

can be written in Ruby 2.0 style as

```ruby
h = { animal: 'cat', vegetable: 'carrot', mineral: 'zeolite' }
```

Outside of this hash literal context, symbols retain their : prefixes wherever they appear in the code. For example, here’s how to get specific values back out of a hash:

```ruby
healthy_snack = h[:vegetable]  # 'carrot'
```

Ruby has an idiosyncratic but powerful convention for handling options in function calls. If a called function requests this behavior, Ruby collects trailing function-call arguments that resemble hash pairs into a new hash. It then passes that hash to the function as an argument. For example, in the Rails expression

```ruby
file_field_tag :upload, accept: 'application/pdf', id: 'commentpdf'
```

the `file_field_tag` receives only two arguments: the `:upload` symbol, and a hash containing the keys `:accept` and `:id`. Because hashes have no inherent order, it doesn’t matter in what order the options appear.

This type of flexible argument processing is a Ruby standard in other ways, too. Ruby libraries, including the standard library, generally do their best to accept the broadest possible range of inputs. Scalars, arrays, and hashes are often equally valid arguments, and many functions can be called with or without blocks.
**Regular expressions in Ruby**

Unlike Python, Ruby has a little bit of language-side sugar to help you deal with regular expressions. Ruby supports the traditional `/.../` notation for regular expression literals, and the contents can include `#{}` escape sequences, much like double-quoted strings.

Ruby also defines the =~ operator (and its negation, !~) to test for a match between a string and a regular expression. It evaluates either to the index of the first match or to nil if there is no match.

```
"Hermann Hesse" =~ /H[aeiou]/  # => 0
```

To access the components of a match, explicitly invoke the regular expression’s match method. It returns either nil (if no match) or an object that can be accessed as an array of components.

```
if m = /(^H\w*)\s/.match("Heinrich Hoffmeyer headed this heist")
  puts m[0]  # 'Heinrich'
end
```

Here’s a look at a Ruby version of the dwarf-suit example from this page:

```ruby
suits = {
  Dopey: 'green', Happy: 'blue', Sleepy: 'taupe
}

abort "Usage: #{ARGV[0]} pattern" unless ARGV.size == 1
pat = /#{ARGV[0]}/;
matches = suits.lazy.select { |dwarf, color| pat =~ dwarf || pat =~ color }
if matches.any?
  dwarf, color = matches.first
  print "%s's dwarf suit is %s\n" %
    [ dwarf.to_s.sub(pat, '_\1_'), color.sub(pat, '_\1_') ]
else
  print "No dwarves or dwarf suits matched the pattern.\n"
end
```

The `select` method on a collection creates a new collection that includes only the elements for which the supplied block evaluates to true. In this case, `matches` is a new hash that includes only pairs for which either the key or the value matches the search pattern. Since we made the starting hash lazy, the filtering won’t actually occur until we try to extract values from the result. In fact, this code checks only as many pairs as are needed to find a match.

Did you notice that the =~ pattern-matching operator was used on the symbols that represent the dwarves’ names? It works because =~ is smart enough to convert the symbols to strings before matching. Unfortunately, we have to perform the conversion explicitly (with the `to_s` method)
when applying the substitution pattern; sub is only defined on strings, so we need a real, live string on which to call it.

Note also the parallel assignment of dwarf and color. matches.first returns a two-element array, which Ruby automatically unpacks.

The % operator for strings works similarly to the same operator in Python; it’s the Ruby version of sprintf. Here there are two components to fill in, so we pass in the values as a two-element array.
Ruby as a filter

You can use Ruby without a script by putting isolated expressions on the command line. This is an easy way to do quick text transformations (though truth be told, Perl is still much better at this role).

Use the -p and -e command-line options to loop through STDIN, run a simple expression on each line (represented as the variable $_), and print the result. For example, the following command translates /etc/passwd to upper case:

```
ruby -pe '$._.tr("a-z", "A-Z")' /etc/passwd
```

```
NOBODY:*:-2:-2:UNPRIVILEGED USER:/VAR/EMPTY:/USR/BIN/FALSE
ROOT:*:0:0:SYSTEM ADMINISTRATOR:/VAR/ROOT:/BIN/SH
...
```

ruby -a turns on autosplit mode, which separates input lines into fields that are stored in the array named $F. Whitespace is the default field separator, but you can set another separator pattern with the -F option.

Autosplit is handy to use in conjunction with -p or its nonautoprinting variant, -n. The command below uses ruby -ane to produce a version of the passwd file that includes only usernames and shells.

```
ruby -F: -ane 'print $F[0], ":", $F[-1]' /etc/passwd
```

```
nobody:/usr/bin/false
root:/bin/sh
...
```

The truly intrepid can use -i in conjunction with -pe to edit files in place; Ruby reads in the files, presents their lines for editing, and saves the results out to the original files. You can supply a pattern to -i that tells Ruby how to back up the original version of each file. For example, -i.bak backs up passwd as passwd.bak. Beware—if you don’t supply a backup pattern, you don’t get backups at all. Note that there’s no space between the -i and the suffix.
7.7 Library and environment management for Python and Ruby

Languages have many of the same packaging and version control issues that operating systems do, and they often resolve them in analogous ways. Python and Ruby are similar in this area, so we discuss them together in this section.
Finding and installing packages

The most basic requirement is some kind of easy and standardized way to discover, obtain, install, update, and distribute add-on software. Both Ruby and Python have centralized warehouses for this purpose, Ruby’s at rubygems.org and Python’s at pypi.python.org.

In the Ruby world, packages are called “gems,” and the command that wrangles them is called `gem` as well. `gem search regex` shows the available gems with matching names, and `gem install gem-name` downloads and installs a gem. You can use the `--user-install` option to install a private copy instead of modifying the system’s complement of gems.

The Python equivalent is called `pip` (or `pip2` or `pip3`, depending on which Python versions are installed). Not all systems include `pip` by default. Those that don’t typically make it available as a separate (OS-level) package. As with `gem`, `pip search` and `pip install` are the mainstay commands. A `--user` option installs packages into your home directory.

Both `gem` and `pip` understand dependencies among packages, at least at a basic level. When you install a package, you’re implicitly asking for all the packages it depends on to be installed as well (if they are not already present).

In a basic Ruby or Python environment, only a single version of a package can be installed at once. If you reinstall or upgrade a package, the old version is removed.

You often have the choice to install a gem or pip package through the standard language mechanism (`gem` or `pip`) or through an OS-level package that’s stocked in your vendor’s standard repository. OS packages are more likely to be installed and run without issues, but they are less likely to be up to date. Neither option is clearly superior.
Creating reproducible environments

Programs, libraries, and languages develop complex webs of dependencies as they evolve together over time. A production-level server might depend on tens or hundreds of these components, each of which has its own expectations about the installation environment. How do you identify which combination of library versions will create a harmonious environment? How do you make sure the configuration you tested in the development lab is the same one that gets deployed to the cloud? More basically, how do you make sure that managing all these parts isn’t a big hassle?

Both Python and Ruby have a standardized way for packages to express their dependencies. In both systems, package developers create a text file at the root of the project that lists its dependencies. For Ruby, the file is called **Gemfile**, and for Python, **requirements.txt**. Both formats support flexible version specifications for dependencies, so it’s possible for packages to declare that they’re compatible with “any release of **simplejson** version 3 or higher” or “Rails 3, but not Rails 4.” It’s also possible to specify an exact version requirement for any dependency.

Both file formats allow a source to be specified for each package, so dependencies need not be distributed through the language’s standard package warehouse. All common sources are supported, from web URLs to local files to GitHub repositories.

You install a batch of Python dependencies with **pip install -r requirements.txt**. Although **pip** does a fine job of resolving individual version specifications, it’s unfortunately not able to solve complex dependency relationships among packages on its own. Developers sometimes have to tweak the order in which packages are mentioned in the **requirements.txt** file to achieve a satisfactory result. It’s also possible, though uncommon, for new package releases to disturb the version equilibrium.

**pip freeze** prints out Python’s current package inventory in **requirements.txt** format, specifying an exact version for each package. This feature can be helpful for replicating the current environment on a production server.

In the Ruby world, **gem install -g Gemfile** is a fairly direct analog of **pip -r**. In most circumstances, though, it’s better to use the Bundler gem to manage dependencies. Run **gem install bundler** to install it (if it’s not already on the system), then run **bundle install** from the root directory of the project you’re setting up. (Ruby gems can include shell-level commands. They don’t typically have man pages, though; run **bundle help** for details, or see bundler.io for complete documentation.)

Bundler has several nice tricks up its sleeve:

- It does true recursive dependency management, so if there’s a set of gems that are mutually compatible and that satisfy all constraints, Bundler can find it without help.
It automatically records the results of version calculations in a file called `Gemfile.lock`. Maintaining this context information lets Bundler handle updates to the `Gemfile` conservatively and efficiently. Bundler modifies only the packages it needs to when migrating to a new version of the `Gemfile`.

Because `Gemfile.lock` is sticky in this way, running `bundle install` on a deployment server automatically reproduces the package environment found in the development environment. (Or at least, that’s the default behavior. It’s easy to specify different requirements for development and deployment environments in the `Gemfile` if you need to.)

In deployment mode (`bundle install --deployment`), Bundler installs missing gems into the local project directory, helping isolate the project from any future changes to the system’s package complement. You can then use `bundle exec` to run specific commands within this hybrid gem environment. Some software packages, such as Rails, are Bundler-aware and will use the locally installed packages even without a `bundle exec` command.
Multiple environments

**pip** and **bundle** handle dependency management for individual Python and Ruby programs, but what if two programs on the same server have conflicting requirements? Ideally, every program in a production environment would have its own library environment that was independent of the system and of all other programs.

**virtualenv: virtual environments for Python**

Python’s **virtualenv** package creates virtual environments that live within their own directories. (As with other Python-related commands, there are numeric-suffixed versions of the **virtualenv** command that go with particular Python versions.) After installing the package, just run the **virtualenv** command with a pathname to set up a new environment:

```
$ virtualenv myproject
New python executable in /home/ulsah/myproject/bin/python
Installing setuptools, pip, wheel...done.
```

Each virtual environment has a **bin/** directory that includes binaries for Python and PIP. When you run one of those binaries, you’re automatically placed in the corresponding virtual environment. Install packages into the environment as usual by running the virtual environment’s copy of **pip**.

To start a virtualized Python program from **cron** or from a system startup script, explicitly specify the path to the proper copy of **python**. (Alternatively, put the path in the script’s shebang line.)

When working interactively in the shell, you can **source** a virtual environment’s **bin/activate** script to set the virtual environment’s versions of **python** and **pip** as the defaults. The script rearranges your shell’s PATH variable. Use **deactivate** to leave the virtual environment.

Virtual environments are tied to specific versions of Python. At the time a virtual environment is created, you can set the associated Python binary with virtualenv’s **--python** option. The Python binary must already be installed and functioning.

**RVM: the Ruby enVironment Manager**

Things are similar in the Ruby world, but somewhat more configurable and more complicated. You saw on this page that Bundler can cache local copies of Ruby gems on behalf of a specific application. This is a reasonable approach when moving projects into production, but it isn’t so great for interactive use. It also assumes that you want to use the system’s installed version of Ruby.

Those who want a more general solution should investigate RVM, a complex and rather unsightly environment virtualizer that uses a bit of shell hackery. To be fair, RVM is an extremely polished example of the “unsightly hack” genus. In practice, it works smoothly.
RVM manages both Ruby versions and multiple gem collections, and it lets you switch among all these on the fly. For example, the command

```
$ rvm ruby-2.3.0@ulsah
```

activates Ruby version 2.3.0 and the gemset called ulsah. References to `ruby` or `gem` now resolve to the specified versions. This magic also works for programs installed by gems, such as `bundle` and `rails`. Best of all, gem management is unchanged; just use `gem` or `bundle` as you normally would, and any newly installed gems automatically end up in the right place.

RVM’s installation procedure involves fetching a Bash script from the web and executing it locally. Currently, the commands are

```
$ curl -o /tmp/install -sSL https://get.rvm.io
$ sudo bash /tmp/install stable
```

but check rvm.io for the current version and a cryptographic signature. (Also see this page for some comments on why our example commands don’t exactly match RVM’s recommendations.) Be sure to install with `sudo` as shown here; if you don’t, RVM sets up a private environment in your home directory. (That works fine, but nothing on a production system should refer to your home directory.) You’ll also need to add authorized RVM users to the rvm UNIX group.

After the initial RVM installation, don’t use `sudo` when installing gems or changing RVM configurations. RVM controls access through membership in the rvm group.

Under the covers, RVM does its magic by manipulating the shell’s environment variables and search path. Ergo, it has to be invited into your environment like a vampire by running some shell startup code at login time. When you install RVM at the system level, RVM drops an `rvm.sh` scriptlet with the proper commands into `/etc/profile.d`. Some shells automatically run this stub. Those that don’t just need an explicit `source` command, which you can add to your shell’s startup files:

```
source /etc/profile.d/rvm.sh
```

RVM doesn’t modify the system’s original Ruby installation in any way. In particular, scripts that start with a

```
#!/usr/bin/ruby
```

shebang continue to run under the system’s default Ruby and to see only system-installed gems. The following variant is more liberal:

```
#!/usr/bin/env ruby
```

It locates the `ruby` command according to the RVM context of the user that runs it.
**rvm install** installs new versions of Ruby. This RVM feature makes it quite painless to install different versions of Ruby, and it should generally be used in preference to your OS’s native Ruby packages, which are seldom up to date. **rvm install** downloads binaries if they are available. If not, it installs the necessary OS packages and then builds Ruby from source code.

Here’s how we might set up for deployment a Rails application known to be compatible with Ruby 2.2.1:

```
$ rvm install ruby-2.2.1
Searching for binary rubies, this might take some time.
No binary rubies available for: ubuntu/15.10/x86_64/ruby-2.2.1.
Continuing with compilation. Please read 'rvm help mount' to get more information on binary rubies.
Checking requirements for ubuntu.
Installing required packages: gawk, libreadline6-dev, zlib1g-dev,
   libncurses5-dev, automake, libtool, bison, libffi-dev..............
Requirements installation successful.
Installing Ruby from source to: /usr/local/rvm/rubies/ruby-2.2.1, this may take a while depending on your cpu(s)...
...
```

If you installed RVM as described above, the Ruby system is installed underneath `/usr/local/rvm` and is accessible to all accounts on the system.

Use **rvm list known** to find out which versions of Ruby RVM knows how to download and build. Rubies shown by **rvm list** have already been installed and are available for use.

```
$ cd myproject.rails
$ rvm ruby-2.2.1@myproject --create --default --ruby-version
ruby-2.2.1 - #gemset created /usr/local/rvm/gems/ruby-2.2.1@myproject
ruby-2.2.1 - #generating myproject wrappers..........
$ gem install bundler
Fetching: bundler-1.11.2.gem (100%)
Successfully installed bundler-1.11.2
1 gem installed
$ bundle
Fetching gem metadata from https://rubygems.org/.........
Fetching version metadata from https://rubygems.org/...
Fetching dependency metadata from https://rubygems.org/..
Resolving dependencies......
...
```

The **ruby-2.2.1@myproject** line specifies both a Ruby version and a gemset. The **--create** flag creates the gemset if it doesn’t already exist. **--default** makes this combination your RVM default, and **--ruby-version** writes the names of the Ruby interpreter and gemset to **.ruby-version** and **.ruby-gemset** in the current directory.
If the .*-version files exist, RVM automatically reads and honors them when dealing with scripts in that directory. This feature allows each project to specify its own requirements and frees you from the need to remember what goes with what.

To run a package in its requested environment (as documented by .ruby-version and .ruby-gemset), run the command

```bash
rvm in /path/to/dir do startup-cmd startup-arg ...
```

This is a handy syntax to use when running jobs out of startup scripts or cron. It doesn’t depend on the current user having set up RVM or on the current user’s RVM configuration.

Alternatively, you can specify an explicit environment for the command, as in

```bash
rvm ruby-2.2.1@myproject do startup-cmd startup-arg ...
```

Yet a third option is to run a ruby binary from within a wrapper maintained by RVM for this purpose. For example, running

```
/usr/local/rvm/wrappers/ruby-2.2.1@myproject/ruby ...
```

automatically transports you into the Ruby 2.2.1 world with the myproject gemset.
7.8 Revision Control with Git

Mistakes are a fact of life. It’s important to keep track of configuration and code changes so that when these changes cause problems, you can easily revert to a known-good state. Revision control systems are software tools that track, archive, and grant access to multiple revisions of files.

Revision control systems address several problems. First, they define an organized way to trace the history of modifications to a file such that changes can be understood in context and so that earlier versions can be recovered. Second, they extend the concept of versioning beyond the level of individual files. Related groups of files can be versioned together, taking into account their interdependencies. Finally, revision control systems coordinate the activities of multiple editors so that race conditions cannot cause anyone’s changes to be permanently lost and so that incompatible changes from multiple editors do not become active simultaneously.

By far the most popular system in use today is Git, created by the one and only Linus Torvalds. Linus created Git to manage the Linux kernel source code because of his frustration with the version control systems that existed at the time. It is now as ubiquitous and influential as Linux. It’s difficult to tell which of Linus’s inventions has had a greater impact on the world.

See this page for more information about DevOps.

Most modern software is developed with help from Git, and as result, administrators encounter it daily. You can find, download, and contribute to open source projects on GitHub, GitLab, and other social development sites. You can also use Git to track changes to scripts, configuration management code, templates, and any other text files that need to be tracked over time. We use Git to track the contents of this book. It’s well suited to collaboration and sharing, making it an essential tool for sites that embrace DevOps.

Git’s shtick is that it has no distinguished central repository. To access a repository, you clone it (including its entire history) and carry it around with you like a hermit crab lugging its shell. Your commits to the repository are local operations, so they’re fast and you don’t have to worry about communicating with a central server. Git uses an intelligent compression system to reduce the cost of storing the entire history, and in most cases this system is quite effective.

Git is great for developers because they can pile their source code onto a laptop and work without being connected to a network while still reaping all the benefits of revision control. When the time comes to integrate multiple developers’ work, their changes can be integrated from one copy of the repository to another in any fashion that suits the organization’s workflow. It’s always possible to unwind two copies of a repository back to their common ancestor state, no matter how many changes and iterations have occurred after the split.
Git’s use of a local repository is a big leap forward in revision control—or perhaps more accurately, it’s a big leap backward, but in a good way. Early revision control systems such as RCS and CVS used local repositories but were unable to handle collaboration, change merging, and independent development. Now we’ve come full circle to a point where putting files under revision control is once again a fast, simple, local operation. At the same time, all Git’s advanced collaboration features are available for use in situations that require them.

Git has hundreds of features and can be quite puzzling in advanced use. However, most Git users get by with only a handful of simple commands. Special situations are best handled by searching Google for a description of what you want to do (e.g., “git undo last commit”). The top result is invariably a Stack Overflow discussion that addresses your exact situation. Above all, don’t panic. Even if it looks like you screwed up the repository and deleted your last few hours of work, Git very likely has a copy stashed away. You just need the reflog fairy to go and fetch it.

Before you start using Git, set your name and email address:

```bash
$ git config --global user.name "John Q. Ulsah"
$ git config --global user.email "ulsah@admin.com"
```

These commands create the ini-formatted Git config file `~/.gitconfig` if it doesn’t already exist. Later `git` commands look in this file for configuration settings. Git power users make extensive customizations here to match their desired workflow.
A simple Git example

We’ve contrived for you a simple example repository for maintaining some shell scripts. In practice, you can use Git to track configuration management code, infrastructure templates, ad hoc scripts, text documents, static web sites, and anything else you need to work on over time.

The following commands create a new Git repository and populate its baseline:

```
$ pwd
/home/bwhaley
$ mkdir scripts && cd scripts
$ git init
Initialized empty Git repository in /home/bwhaley/scripts/.git/
$ cat > super-script.sh << EOF
> #!/bin/sh
> echo "Hello, world"
> EOF
$ chmod +x super-script.sh
$ git add .
$ git commit -m "Initial commit"
[master (root-commit) 9a4d90c] super-script.sh
 create mode 100755 super-script.sh
 1 file changed, 0 insertions(+), 0 deletions(-)
```

In the sequence above, `git init` creates the repository’s infrastructure by creating a `.git` directory in `/home/bwhaley/scripts`. Once you set up an initial “hello, world” script, the command `git add .` copies it to Git’s “index,” which is a staging area for the upcoming commit.

The index is not just a list of files to commit; it’s a bona fide file tree that’s every bit as real as the current working directory and the contents of the repository. Files in the index have contents, and depending on what commands you run, those contents may end up being different from both the repository and the working directory. `git add` really just means “`cp` from the working directory to the index.”

`git commit` enters the contents of the index into the repository. Every commit needs a log message. The `-m` flag lets you include the message on the command line. If you leave it out, `git` starts up an editor for you.

Now make a change and check it into the repository.

```
$ vi super-script.sh
$ git commit super-script.sh -m "Made the script more super"
[master 67514f1] Made the script more super
 1 file changed, 1 insertions(+), 0 deletions(-)
```

Naming the modified files on the `git commit` command line bypasses Git’s normal use of the index and creates a revision that includes only changes to the named files. The existing index
remains unchanged, and Git ignores any other files that may have been modified.

If a change involves multiple files, you have a couple of options. If you know exactly which files were changed, you can always list them on the command line as shown above. If you’re lazy, you can run `git commit -a` to make Git add all modified files to the index before doing the commit. This last option has a couple of pitfalls, however.

First, there may be modified files that you don’t want to include in the commit. For example, if `super-script.sh` had a config file and you had modified that config file for debugging, you might not want to commit the modified file back to the repository.

The second issue is that `git commit -a` picks up only changes to files that are currently under revision control. It does not pick up new files that you may have created in the working directory.

For an overview of Git’s state, you can run `git status`. This command informs you of new files, modified files, and staged files all at once. For example, suppose that you added `more-scripts/another-script.sh`. Git might show the following:

```
$ git status
On branch master
Changes not staged for commit:
  (use "git add <file>..." to update what will be committed)
  (use "git checkout -- <file>..." to discard changes in working directory)

    modified:   super-script.sh

Untracked files:
  (use "git add <file>..." to include in what will be committed)

    more-scripts/
    tmpfile

no changes added to commit (use "git add" and/or "git commit -a")
```

`another-script.sh` is not listed by name because Git doesn’t yet see beneath the `more-scripts` directory that contains it. You can see that `super-script.sh` has been modified, and you can also see a spurious `tmpfile` that probably shouldn’t be included in the repository. You can run `git diff super-script.sh` to see the changes made to the script. `git` helpfully suggests commands for the next operations you may want to perform.

Suppose you want to track the changes to `super-script.sh` separately from your new `another-script.sh`.

```
$ git commit super-script.sh -m "The most super change yet"
Created commit 6f7853c: The most super change yet
  1 files changed, 1 insertions(+), 0 deletions(-)
```

To eradicate `tmpfile` from Git’s universe, create or edit a `.gitignore` file and put the filename inside it. This makes Git ignore the `tmpfile` now and forever. Patterns work, too.
Finally, commit all the outstanding changes:

```
$ echo tmpfile >> .gitignore

$ git add .
$ sudo git commit -m "Ignore tmpfile; Add another-script.sh to the repo"
```

Created commit 32978e6: Ignore tmpfile; add another-script.sh to the repo
2 files changed, 2 insertions(+), 0 deletions(-)
create mode 100644 .gitignore
create mode 100755 more-scripts/another-script.sh

Note that the .gitignore file itself becomes part of the managed set of files, which is usually what you want. It’s fine to re-add files that are already under management, so git add . is an easy way to say “I want to make the new repository image look like the working directory minus anything listed in .gitignore.” You couldn’t just do a git commit -a in this situation because that would pick up neither another-script.sh nor .gitignore; these files are new to Git and so must be explicitly added.
Git caveats

In an effort to fool you into thinking that it manages files’ permissions as well as their contents, Git shows you file modes when adding new files to the repository. It’s lying; Git does not track modes, owners, or modification times.

Git does track the executable bit. If you commit a script with the executable bit set, any future clones will also be executable. But don’t expect Git to track ownership or read-only status. A corollary is that you can’t count on using Git to recover complex file hierarchies in situations where ownerships and permissions are important.

Another corollary is that you should never include plain text passwords or other secrets in a Git repository. Not only are they open to inspection by anyone with access to the repository, but they may also be inadvertently unpacked in a form that’s accessible to the world.
Social coding with Git

The emergence and rapid growth of social development sites such as GitHub and GitLab is one of the most important trends in recent computing history. Millions of open source software projects are built and managed transparently by huge communities of developers who use every conceivable language. Software has never been easier to create and distribute.

GitHub and GitLab are, in essence, hosted Git repositories with a lot of added features that relate to communication and workflow. Anyone can create a repository. Repositories are accessible both through the `git` command and on the web. The web UI is friendly and offers features to support collaboration and integration.

The social coding experience can be somewhat intimidating for neophytes, but in fact it isn’t complicated once some basic terms and methodology are understood.

- “master” is the default name assigned to the first branch in a new repository. Most software projects use this default as their main line of development, although some may not have a master branch at all. The master branch is usually managed to contain current but functional code; bleeding-edge development happens elsewhere. The latest commit is known as the tip or head of the master branch.

- On GitHub, a fork is a snapshot of a repository at a specific point in time. Forks happen when a user doesn’t have permission to modify the main repository but wants to make changes, either for future integration with the primary project or to create an entirely separate development path.

- A pull request is a request to merge changes from one branch or fork to another. They’re read by the maintainers of the target project and can be accepted to incorporate code from other users and developers. Every pull request is also a discussion thread, so both principals and kibitzers can comment on prospective code updates.

- A committer or maintainer is an individual who has write access to a repository. For large open source projects, this highly coveted status is given only to trusted developers who have a long history of contributions.

You’ll often land in a GitHub or GitLab repository when trying to locate or update a piece of software. Make sure you’re looking at the trunk repository and not some random person’s fork. Looked for a “forked from” indication and follow it.

Be cautious when evaluating new software from these sites. Below are a few questions to ponder before rolling out a random piece of new software at your site:

- How many contributors have participated in development?
Does the commit history indicate recent, regular development?

What is the license, and is it compatible with your organization’s needs?

What language is the software written in, and do you know how to manage it?

Is the documentation complete enough for effective use of the software?

Most projects have a particular branching strategy that they rely on to track changes to the software. Some maintainers insist on rigorous enforcement of their chosen strategy, and others are more lenient. One of the most widely used is the Git Flow model developed by Vincent Driessen; see goo.gl/GDaF for details. Before contributing to a project, familiarize yourself with its development practices to help out the maintainers.

Above all, remember that open source developers are often unpaid. They appreciate your patience and courtesy when engaging through code contributions or opening support issues.
7.9 **RECOMMENDED READING**


The complete Pro Git book, released for free under a Creative Commons license.
Shells and shell scripting


POWERS, SHELLEY, JERRY PEEK, TIM O’REILLY, AND MIKE LOUKIDES. *Unix Power Tools, (3rd Edition)*, Sebastopol, CA: O’Reilly Media, 2002. This classic UNIX book covers a lot of ground, including sh scripting and various feats of command-line-fu. Some sections are not aging gracefully, but the shell-related material remains relevant.


BLUM, RICHARD, AND CHRISTINE BRESNANAHAN. *Linux Command Line and Shell Scripting Bible (3rd Edition)*. Indianapolis, IN: John Wiley & Sons, Inc. 2015. This book focuses a bit more specifically on the shell than does the Shotts book, though it’s also bash-specific.

Regular expressions


GOYVAERTS, JAN. regular-expressions.info. A detailed on-line source of information about regular expressions in all their various dialects.

**Python**

**SWEIGART, AL.** *Automate the Boring Stuff with Python: Practical Programming for Total Beginners*. San Francisco, CA: No Starch Press, 2015. This is an approachable introductory text for Python 3 and programming generally. Examples include common administrative tasks.


Ruby

FLANAGAN, DAVID, AND YUKIHIRO MATSUMOTO. *The Ruby Programming Language*. Sebastopol, CA: O’Reilly Media, 2008. This classic, concise, and well-written summary of Ruby comes straight from the horse’s mouth. It’s relatively matter-of-fact and does not cover Ruby 2.0 and beyond; however, the language differences are minor.

BLACK, DAVID A. *The Well-Grounded Rubyist (2nd Edition)*. Shelter Island, NY: Manning Publications, 2014. Don’t let the title scare you off if you don’t have prior Ruby experience; this is a good, all-around introduction to Ruby 2.1.


Modern computing environments span physical hardware, cloud systems, and virtual hosts. Along with the flexibility of this hybrid infrastructure comes an increasing need for centralized and structured account management. System administrators must understand both the traditional account model used by UNIX and Linux and the ways in which this model has been extended to integrate with directory services such as LDAP and Microsoft’s Active Directory.

Account hygiene is a key determinant of system security. Infrequently used accounts are prime targets for attackers, as are accounts with easily guessed passwords. Even if you use your system’s automated tools to add and remove users, it’s important to understand the changes the tools are making. For this reason, we start our discussion of account management with the flat files you would modify to add users to a stand-alone machine. In later sections, we examine the higher-level user management commands that come with our example operating systems and the configuration files that control their behavior.

Most systems also have simple GUI tools for adding and removing users, but these tools don’t usually support advanced features such as a batch mode or advanced localization. The GUI tools are simple enough that we don’t think it’s helpful to review their operation in detail, so in this chapter we stick to the command line.

This chapter focuses fairly narrowly on adding and removing users. Many topics associated with user management actually live in other chapters and are referenced here only indirectly. For example,
• Pluggable authentication modules (PAM) for password encryption and the enforcement of strong passwords are covered in Chapter 17, Single Sign-On. See the material starting on this page.

• Password vaults for managing passwords are described in Chapter 27, Security (see this page).

• Directory services such as OpenLDAP and Active Directory are outlined in Chapter 17, Single Sign-On, starting here.

• Policy and regulatory issues are major topics of Chapter 31, Methodology, Policy, and Politics.
8.1 Account Mechanics

A user is really nothing more than a number. Specifically, an unsigned 32-bit integer known as the user ID or UID. Almost everything related to user account management revolves around this number.

The system defines an API (through standard C library routines) that maps UID numbers back and forth into more complete sets of information about users. For example, `getpwuid()` accepts a UID as an argument and returns a corresponding record that includes information such as the associated login name and home directory. Likewise, `getpwnam()` looks up this same information by login name.

Traditionally, these library calls obtained their information directly from a text file, `/etc/passwd`. As time went on, they began to support additional sources of information such as network information databases (e.g., LDAP) and read-protected files in which encrypted passwords could be stored more securely.

---

See this page for more details regarding the `nsswitch.conf` file.

---

These layers of abstraction (which are often configured in the `nsswitch.conf` file) enable higher-level processes to function without direct knowledge of the underlying account management method in use. For example, when you log in as “dotty”, the logging-in process (window server, `login`, `getty`, or whatever) does a `getpwnam()` on dotty and then validates the password you supply against the encrypted passwd record returned by the library, regardless of its actual origin.

We start with the `/etc/passwd` file approach, which is still supported everywhere. The other options emulate this model in spirit if not in form.
8.2 THE /etc/passwd FILE

/etc/passwd is a list of users recognized by the system. It can be extended or replaced by one or more directory services, so it’s complete and authoritative only on stand-alone systems.

Historically, each user’s encrypted password was also stored in the /etc/passwd file, which is world-readable. However, the onset of more powerful processors made it increasingly feasible to crack these exposed passwords. In response, UNIX and Linux moved the passwords to a separate file (/etc/master.passwd on FreeBSD and /etc/shadow on Linux) that is not world-readable. These days, the passwd file itself contains only a pro-forma entry to mark the former location of the password field (x on Linux and * on FreeBSD).

The system consults /etc/passwd at login time to determine a user’s UID and home directory, among other things. Each line in the file represents one user and contains seven fields separated by colons:

- Login name
- Encrypted password placeholder (see this page)
- UID (user ID) number
- Default GID (group ID) number
- Optional “GECOS” information: full name, office, extension, home phone
- Home directory
- Login shell

For example, the following lines are all valid /etc/passwd entries:

```
root:x:0:0:The System,,x6096,:/:/bin/sh
j1:!!:100:0:Jim Lane,ECOT8-3,,:/staff/jl:/bin/sh
dotty:x:101:20::/home/dotty:/bin/tcsh
```

See this page for more information about the nsswitch.conf file.

If user accounts are shared through a directory service such as LDAP, you might see special entries in the passwd file that begin with + or -. These entries tell the system how to integrate the directory service’s data with the contents of the passwd file. This integration can also be set up in the /etc/nsswitch.conf file.

The following sections discuss the /etc/passwd fields in more detail.
Login name

Login names (also known as usernames) must be unique and, depending on the operating system, may have character set restrictions. All UNIX and Linux flavors currently limit logins to 32 characters.

Login names can never contain colons or newlines, because these characters are used as field separators and entry separators in the `passwd` file, respectively. Depending on the system, other character restrictions may also be in place. Ubuntu is perhaps the most lax, as it allows logins starting with—or consisting entirely of—numbers and other special characters. For reasons too numerous to list, we recommend sticking with alphanumeric characters for logins, using lower case, and starting login names with a letter.

Login names are case sensitive. We are not aware of any problems caused by mixed-case login names, but lowercase names are traditional and also easier to type. Confusion could ensue if the login names john and John were different people.

Login names should be easy to remember, so random sequences of letters do not make good login names. Since login names are often used as email addresses, it’s useful to establish a standard way of forming them. It should be possible for users to make educated guesses about each other’s login names. First names, last names, initials, or some combination of these make reasonable naming schemes.

Keep in mind that some email systems treat addresses as being case insensitive, which is yet another good reason to standardize on lowercase login names. RFC5321 requires that the local portion of an address (that is, the part before the @ sign) be treated as case sensitive. The remainder of the address is handled according to the standards of DNS, which is case insensitive. Unfortunately, this distinction is subtle, and it is not universally implemented. Remember also that many legacy email systems predate the authority of the IETF.

Any fixed scheme for choosing login names eventually results in duplicate names, so you sometimes have to make exceptions. Choose a standard way of dealing with conflicts, such as adding a number to the end.

It’s common for large sites to implement a full-name email addressing scheme (e.g., John.Q.Public@mysite.com) that hides login names from the outside world. This is a good idea, but it doesn’t obviate any of the naming advice given above. If for no other reason than the sanity of administrators, it’s best if login names have a clear and predictable correspondence to users’ actual names.

Finally, a user should have the same login name on every machine. This rule is mostly for convenience, both yours and the user’s.
Encrypted password

Historically, systems encrypted users’ passwords with DES. As computing power increased, those passwords became trivial to crack. Systems then moved to hidden passwords and to MD5-based cryptography. Now that significant weaknesses have been discovered in MD5, salted SHA-512-based password hashes have become the current standard. See the Guide to Cryptography document at owasp.org for up-to-date guidance.

Our example systems support a variety of encryption algorithms, but they all default to SHA-512. You shouldn’t need to update the algorithm choice unless you are upgrading systems from much older releases.

- On FreeBSD, the default algorithm can be modified through the `/etc/login.conf` file.
- On Debian and Ubuntu, the default was formerly managed through `/etc/login.defs`, but this practice has since been obsoleted by Pluggable Authentication Modules (PAM). Default password policies, including the hashing algorithm to use, can be found in `/etc/pam.d/common-passwd`.
- On Red Hat and CentOS, the password algorithm can still be set in `/etc/login.defs` or through the `authconfig` command, as shown here:

```
$ sudo authconfig --passalgo=sha512 --update
```

Changing the password algorithm does not update existing passwords, so users must manually update their passwords before the new algorithm can take effect. To invalidate a user’s password and force an update, use

```
$ chage -d 0 username
```

Password quality is another important issue. In theory, longer passwords are more secure, as are passwords that include a range of different character types (e.g., uppercase letters, punctuation marks, and numbers).

Most systems let you impose password construction standards on your users, but keep in mind that users can be adept at skirting these requirements if they find them excessive or burdensome. Table 8.1 shows the default standards used by our example systems.

Table 8.1: Password quality standards
See this page for more comments on password selection.

<table>
<thead>
<tr>
<th>System</th>
<th>Default requirements</th>
<th>Where set</th>
</tr>
</thead>
<tbody>
<tr>
<td>Red Hat</td>
<td>8+ characters, complexity enforced</td>
<td>/etc/login.defs</td>
</tr>
<tr>
<td>CentOS</td>
<td></td>
<td>/etc/security/pwquality.conf</td>
</tr>
<tr>
<td></td>
<td></td>
<td>/etc/pam.d/system-auth</td>
</tr>
<tr>
<td>Debian</td>
<td>6+ characters, complexity enforced</td>
<td>/etc/login.defs</td>
</tr>
<tr>
<td>Ubuntu</td>
<td></td>
<td>/etc/pam.d/common-password</td>
</tr>
<tr>
<td>FreeBSD</td>
<td>No constraints</td>
<td>/etc/login.conf</td>
</tr>
</tbody>
</table>

Password quality requirements are a matter of debate, but we recommend that you prioritize length over complexity (see xkcd.com/comics/password_strength.png). Twelve characters is the minimal length for a future-proof password; note that this is significantly longer than any system’s default. Your site may also have organization-wide standards for password quality. If it does, defer to those settings.

If you choose to bypass your system’s tools for adding users and instead modify /etc/passwd by hand (by running the vipw command—see this page) to create a new account, put a * (FreeBSD) or an x (Linux) in the encrypted password field. This measure prevents unauthorized use of the account until you or the user has set a real password.

Encrypted passwords are of constant length (86 characters for SHA-512, 34 characters for MD5, and 13 characters for DES) regardless of the length of the unencrypted password. Passwords are encrypted in combination with a random “salt” so that a given password can correspond to many different encrypted forms. If two users happen to select the same password, this fact usually cannot be discovered by inspection of the encrypted passwords.

MD5-encrypted password fields in the shadow password file always start with $1$ or $md5$. Blowfish passwords start with $2$, SHA-256 passwords with $5$, and SHA-512 passwords with $6$. 
UID (user ID) number

See this page for a description of the root account.

By definition, root has UID 0. Most systems also define pseudo-users such as bin and daemon to be the owners of commands or configuration files. It’s customary to put such fake logins at the beginning of the /etc/passwd file and to give them low UIDs and a fake shell (e.g., /bin/false) to prevent anyone from logging in as those users.

To allow plenty of room for nonhuman users you might want to add in the future, we recommend that you assign UIDs to real users starting at 1000 or higher. (The desired range for new UIDs can be specified in the configuration files for useradd.) By default, our Linux reference systems start UIDs at 1000 and go up from there. FreeBSD starts the first user at UID 1001 and then adds one for each additional user.

Do not recycle UIDs, even when users leave your organization and you delete their accounts. This precaution prevents confusion if files are later restored from backups, where users may be identified by UID rather than by login name.

UIDs should be kept unique across your entire organization. That is, a particular UID should refer to the same login name and the same person on every machine that person is authorized to use. Failure to maintain distinct UIDs can result in security problems with systems such as NFS and can also result in confusion when a user moves from one workgroup to another.

It can be hard to maintain unique UIDs when groups of machines are administered by different people or organizations. The problems are both technical and political. The best solution is to have a central database or directory server that contains a record for each user and enforces uniqueness.

A simpler scheme is to assign each group within an organization its own range of UIDs and to let each group manage its own range. This solution keeps the UID spaces separate but does not address the parallel issue of unique login names. Regardless of your scheme, consistency of approach is the primary goal. If consistency isn’t feasible, UID uniqueness is the second-best target.

The Lightweight Directory Access Protocol (LDAP) is a popular system for managing and distributing account information and works well for large sites. It is briefly outlined in this chapter starting here and is covered more thoroughly in Chapter 17, Single Sign-On, starting here.
Default GID (group ID) number

Like a UID, a group ID number is a 32-bit integer. GID 0 is reserved for the group called root, system, or wheel. As with UIDs, the system uses several predefined groups for its own housekeeping. Alas, there is no consistency among vendors. For example, the group “bin” has GID 1 on Red Hat and CentOS, GID 2 on Ubuntu and Debian, and GID 7 on FreeBSD.

In ancient times, when computing power was expensive, groups were used for accounting purposes so that the right department could be charged for your seconds of CPU time, minutes of login time, and kilobytes of disk used. Today, groups are used primarily to share access to files.

---

See [this page](#) for more information about setgid directories.

---

The `/etc/group` file defines the groups, with the GID field in `/etc/passwd` providing a default (or “effective”) GID at login time. The default GID is not treated specially when access is determined; it is relevant only to the creation of new files and directories. New files are normally owned by your effective group; to share files with others in a project group, you must manually change the files’ group owner.

To facilitate collaboration, you can set the setgid bit (02000) on a directory or mount filesystems with the `grpid` option. Both of these measures make newly created files default to the group of their parent directory.
GECOS field

The GECOS field is sometimes used to record personal information about each user. The field is a relic from a much earlier time when some early UNIX systems used General Electric Comprehensive Operating Systems for various services. It has no well-defined syntax. Although you can use any formatting conventions you like, conventionally, comma-separated GECOS entries are placed in the following order:

- Full name (often the only field used)
- Office number and building
- Office telephone extension
- Home phone number

See this page for more information about LDAP.

The `chfn` command lets users change their own GECOS information. `chfn` is useful for keeping things like phone numbers up to date, but it can be misused. For example, a user can change the information to be obscene or incorrect. Some systems can be configured to restrict which fields `chfn` can modify; most college campuses disable it entirely. On most systems, `chfn` understands only the `passwd` file, so if you use LDAP or some other directory service for login information, `chfn` may not work at all.
Home directory

A user’s home directory is his or her default directory at login time. Home directories are where login shells look for account-specific customizations such as shell aliases and environment variables, as well as SSH keys, server fingerprints, and other program state.

Be aware that if home directories are mounted over a network filesystem, they may be unavailable in the event of server or network problems. If the home directory is missing at login time, the system might print a message such as “no home directory” and put the user in /.

Such a message appears when you log in on the console or on a terminal, but not when you log in through a display manager such as xdm, gdm, or kdm. Not only will you not see the message, but you will generally be logged out immediately because of the display manager’s inability to write to the proper directory (e.g., ~/.gnome).

Alternatively, the system might disallow home-directory-less logins entirely, depending on the configuration.

Home directories are covered in more detail here.
Login shell

See this page for more information about shells.

The login shell is normally a command interpreter, but it can be any program. A Bourne-shell compatible `sh` is the default for FreeBSD, and `bash` (the GNU “Bourne again” shell) is the default for Linux.

Some systems permit users to change their shell with the `chsh` command, but as with `chfn`, this command might not work if you are using LDAP or some other directory service to manage login information. If you use the `/etc/passwd` file, a sysadmin can always change a user’s shell by editing the `passwd` file with `vipw`. 
8.3 The Linux /etc/shadow File

On Linux, the shadow password file is readable only by the superuser and serves to keep encrypted passwords safe from prying eyes and password cracking programs. It also includes some additional account information that wasn’t provided for in the original /etc/passwd format. These days, shadow passwords are the default on all systems.

The shadow file is not a superset of the passwd file, and the passwd file is not generated from it. You must maintain both files or use tools such as useradd that maintain both files on your behalf. Like /etc/passwd, /etc/shadow contains one line for each user. Each line contains nine fields, separated by colons:

- Login name
- Encrypted password
- Date of last password change
- Minimum number of days between password changes
- Maximum number of days between password changes
- Number of days in advance to warn users about password expiration
- Days after password expiration that account is disabled
- Account expiration date
- A field reserved for future use which is currently always empty

Only the values for the username and password are required. Absolute date fields in /etc/shadow are specified in terms of days (not seconds) since Jan 1, 1970, which is not a standard way of reckoning time on UNIX or Linux systems. To convert between the date in seconds and in days, run `expr `date+%s` / 86400`.

A typical shadow entry looks like this:

```
miller:$6$iTEFbMTMSXmPwEefb9RUBv1zv8EgXQdaZg2e0d5uXylv4sFzi 6G4I1qavLi1TQgniAHm3Czw/LoaGzoFzaMm.YwOl/:16971:0:100:14:::
```

Here is a more complete description of each field:

- The login name is the same as in /etc/passwd. This field connects a user’s passwd and shadow entries.
- The encrypted password is identical in concept and execution to the one previously stored in /etc/passwd.
• The last change field records the time at which the user’s password was last changed. This field is filled in by the `passwd` command.

• The fourth field sets the number of days that must elapse between password changes. The idea is to force authentic changes by preventing users from immediately reverting to a familiar password after a required change. However, this feature can be somewhat dangerous in the aftermath of a security intrusion. We suggest setting this field to 0.

• The fifth field sets the maximum number of days allowed between password changes. This feature allows the administrator to enforce password aging; see this page for more information. Under Linux, the actual enforced maximum number of days is the sum of this field and the seventh (grace period) field.

• The sixth field sets the number of days before password expiration when `login` should begin to warn the user of the impending expiration.

• The eighth field specifies the day (in days since Jan 1, 1970) on which the user’s account will expire. The user cannot log in after this date until the field has been reset by an administrator. If the field is left blank, the account never expires.

You can use `usermod` to set the expiration field. It accepts dates in the format `yyyy-mm-dd`.

• The ninth field is reserved for future use (or at this rate, may never be used).

Let’s look again at our example `shadow` line:

```
millert:$6$iTEFbMTM$CXmxPwErbEef9RUBvf1zvBEG9XQdaZg2e0d5uXyvt4sFzi
6G4lIqavLlTQgniAHm3Czw/LoaGzoFzaMm.YwO1/:17336:0:180:14:::
```

In this example, the user millert last changed his password on June 19, 2017. The password must be changed again within 180 days, and millert will receive warnings that the password needs to be changed for the last two weeks of this period. The account does not have an expiration date.

Use the `pwconv` utility to reconcile the contents of the `shadow` file and those of the `passwd` file, picking up any new additions and deleting users that are no longer listed in `passwd`. 
8.4 FreeBSD's /etc/master.passwd and /etc/login.conf files

The adoption of PAM and the availability of similar user management commands on FreeBSD and Linux have made account administration relatively consistent across platforms, at least at the topmost layer. However, a few differences do exist in the underlying implementation.
The /etc/master.passwd file

On FreeBSD, the “real” password file is `/etc/master.passwd`, which is readable only by root. The `/etc/passwd` file exists for backward compatibility and does not contain any passwords (instead, it has * characters as placeholders).

To edit the password file, run the `vipw` command. This command invokes your editor on a copy of `/etc/master.passwd`, then installs the new version and regenerates the `/etc/passwd` file to reflect any changes. (`vipw` is standard on all UNIX and Linux systems, but it’s particularly important to use on FreeBSD because the dual password files need to stay synchronized. See this page.)

In addition to containing all the fields of the `passwd` file, the `master.passwd` file contains three bonus fields. Unfortunately, they’re squeezed in between the default GID field and the GECOS field, so the file formats are not directly compatible. The extra three fields are

- Login class
- Password change time
- Expiration time

The login class (if one is specified) refers to an entry in the `/etc/login.conf` file. The class determines resource consumption limits and controls a variety of other settings. See the next section for specifics.

The password change time field implements password aging. It contains the time in seconds since the UNIX epoch after which the user will be forced to change his or her password. You can leave this field blank, indicating that the password never expires.

The account expiration time gives the time and date (in seconds, as for password expiration) at which the user’s account will expire. The user cannot log in after this date unless the field is reset by an administrator. If this field is left blank, the account will not expire.
The /etc/login.conf file

FreeBSD’s /etc/login.conf file sets account-related parameters for users and groups of users. Its format consists of colon-delimited key/value pairs and Boolean flags.

When a user logs in, the login class field of /etc/master.passwd determines which entry in /etc/login.conf to apply. If the user’s master.passwd entry does not specify a login class, the default class is used.

A login.conf entry can set any of the following:

- Resource limits (maximum process size, maximum file size, number of open files, etc.)
- Session accounting limits (when logins are allowed, and for how long)
- Default environment variables
- Default paths (PATH, MANPATH, etc.)
- Location of the “message of the day” file
- Host and TTY-based access control
- Default umask
- Account controls (mostly superseded by the PAM module pam_passwdqc)

The following example overrides several of the default values. It’s intended for assignment to system administrators.

```
sysadmin:\n  :ignorenologin: \n  :requirehome:@: \n  :maxproc=unlimited: \n  :openfiles=unlimited: \n  :tc=default:
```

Users in the sysadmin login class are allowed to log in even when /var/run/nologin exists, and they need not have a working home directory (this option permits logins when NFS is not working). Sysadmin users can start any number of processes and open any number of files (no artificial limit is imposed). The last line pulls in the contents of the default entry.

Although FreeBSD has reasonable defaults, you might be interested in updating the /etc/login.conf file to set idle timeout and password expiration warnings. For example, to set the idle timeout to 15 minutes and enable warnings seven days before passwords expire, you would add the following clauses to the definition of default:
When you modify the `/etc/login.conf` file, you must also run the following command to compile your changes into the hashed version of the file that the system actually refers to in daily operation:

```
$ cap_mkdb /etc/login.conf
```
8.5 The `/etc/group` file

The `/etc/group` file contains the names of UNIX groups and a list of each group’s members. Here’s a portion of the `group` file from a FreeBSD system:

```
wheel:*:0:root
sys:*:3:root,bin
operator:*:5:root
bin:*:7:root
ftp:*:14:dan
staff:*:20:dan,ben,trent
nobody:*:65534:1pd
```

Each line represents one group and contains four fields:

- Group name
- Encrypted password or a placeholder
- GID number
- List of members, separated by commas (be careful not to add spaces)

As in `/etc/passwd`, fields are separated by colons. Group names should be limited to eight characters for compatibility, although many systems do not actually require this.

It’s possible to set a group password that allows arbitrary users to enter the group with the `newgrp` command. However, this feature is rarely used. The group password can be set with `gpasswd`, which under Linux stores the encrypted password in the `/etc/gshadow` file.

As with usernames and UIDs, group names and GIDs should be kept consistent among machines that share files through a network filesystem. Consistency can be hard to maintain in a heterogeneous environment because different operating systems use different GIDs for standard system groups.

If a user defaults to a particular group in `/etc/passwd` but does not appear to be in that group according to `/etc/group`, `/etc/passwd` wins the argument. The group memberships granted at login time are the union of those found in the `passwd` and `group` files.

Some older systems limit the number of groups a user can belong to. There is no real limit on current Linux and FreeBSD kernels.

Much as with UIDs, we recommend minimizing the potential for GID collisions by starting local groups at GID 1000 or higher.

The UNIX tradition was originally to add new users to a group that represented their general category such as “students” or “finance.” However, this convention increases the likelihood that
users will be able to read one another’s files because of slipshod permission settings, even if that is not really the intention of the files’ owner.

To avoid this problem, system utilities such as `useradd` and `adduser` now default to putting each user in his or her own personal group (that is, a group named after the user and which includes only that user). This convention is much easier to maintain if personal groups’ GIDs match their corresponding users’ UIDs.

To let users share files by way of the group mechanism, create separate groups for that purpose. The idea behind personal groups is not to discourage the use of groups per se—it’s simply to establish a more restrictive `default` group for each user so that files are not inadvertently shared. You can also limit access to newly created files and directories by setting your user’s default `umask` in a default startup file such as `/etc/profile` or `/etc/bashrc` (see this page).

Group membership can also serve as a marker for other contexts or privileges. For example, rather than entering the username of each system administrator into the `sudoers` file, you can configure `sudo` so that everyone in the “admin” group automatically has `sudo` privileges.

Linux supplies the `groupadd`, `groupmod`, and `groupdel` commands to create, modify, and delete groups.

FreeBSD uses the `pw` command to perform all these functions. To add the user “dan” to the group “staff” and then verify that the change was properly implemented, you would run the following commands:

```
$ sudo pw groupmod staff -m dan
$ pw groups show staff
staff:*:20:dan, evi, garth, trent, ben
```
8.6 Manual steps for adding users

Before you create an account for a new user at a corporate, government, or educational site, it’s important that the user sign and date a copy of your local user agreement and policy statement. (What?! You don’t have a user agreement and policy statement? See this page for more information about why you need one and what to put in it.)

Users have no particular reason to want to sign a policy agreement, so it’s to your advantage to secure their signatures while you still have some leverage. We find that it takes extra effort to secure a signed agreement after an account has been released. If your process allows for it, have the paperwork precede the creation of the account.

Mechanically, the process of adding a new user consists of several steps required by the system and a few more that establish a useful environment for the new user and incorporate the user into your local administrative system.

Required:

- Edit the passwd and shadow files (or the master.passwd file on FreeBSD) to define the user’s account.
- Add the user to the /etc/group file (not really necessary, but nice).
- Set an initial password.
- Create, chown, and chmod the user’s home directory.
- Configure roles and permissions (if you use RBAC; see this page).

For the user:

- Copy default startup files to the user’s home directory.

For you:

- Have the new user sign your policy agreement.
- Verify that the account is set up correctly.
- Document the user’s contact information and account status.

This list cries out for a script or tool, and fortunately, each of our example systems includes at least a partial off-the-shelf solution in the form of an adduser or useradd command. We take a look at these tools starting on this page.
Editing the passwd and group files

Manual maintenance of the passwd and group files is error prone and inefficient, so we recommend the slightly higher-level tools such as useradd, adduser, usermod, pw, and chsh as daily drivers.

If you do have to make manual changes, use the vipw command to edit the passwd and shadow files (or on FreeBSD, the master.passwd file). Although it sounds vi-centric, it actually invokes your favorite editor as defined in the EDITOR environment variable. More importantly, it locks the files so that editing sessions (or your editing and a user’s password change) cannot collide.

After you run vipw, our Linux reference systems remind you to edit the shadow file after you have edited the passwd file. Use vipw -s to do so.

Under FreeBSD, vipw edits the master.passwd file instead of /etc/passwd. After installing your changes, vipw runs pwd_mkdb to generate the derived passwd file and two hashed versions of master.passwd (one that contains the encrypted passwords and is readable only by root, and another that lacks the passwords and is world-readable).

For example, running vipw and adding the following line would define an account called whitney:

```
whitney:*:1003:1003::0:0:Whitney Sather, AMATH 3-27, x7919,: 
   /home/staff/whitney:/bin/sh
```

Note the star in the encrypted password field. This prevents use of the account until a real password is set with the passwd command (see the next section).

Next, edit /etc/group by running vigr. Add a line for the new personal group if your site uses them, and add the user's login name to each of the groups in which the user should have membership.

As with vipw, using vigr ensures that the changes made to the /etc/group file are sane and atomic. After an edit session, vigr should prompt you to run vigr -s to edit the group shadow (gshadow) file as well. Unless you want to set a password for the group—which is unusual—you can skip this step.

On FreeBSD, use pw groupmod to make changes to the /etc/group file.
Setting a password

Set a password for a new user with

$ sudo passwd newusername

You’ll be prompted for the actual password.

Some automated systems for adding new users do not require you to set an initial password. Instead, they force the user to set a password on first login. Although this feature is convenient, it’s a giant security hole: anyone who can guess new login names (or look them up in `/etc/passwd`) can swoop down and hijack accounts before the intended users have had a chance to log in.

Among many other functions, FreeBSD’s `pw` command can also generate and set random user passwords:

$ sudo pw usermod raphael -w random
    Password for 'raphael' is: 1n3tcYu1s

See `this page` for tips on selecting good passwords.

We’re generally not fans of random passwords for ongoing use. However, they are a good option for transitional passwords that are only intended to last until the account is actually used.
Creating the home directory and installing startup files

`useradd` and `adduser` create new users’ home directories for you, but you’ll likely want to double-check the permissions and startup files for new accounts.

There’s nothing magical about home directories. If you neglected to include a home directory when setting up a new user, you can create it with a simple `mkdir`. You need to set ownerships and permissions on the new directory as well, but this is most efficiently done after you’ve installed any local startup files.

Startup files traditionally begin with a dot and end with the letters `rc`, short for “run command,” a relic of the CTSS operating system. The initial dot causes `ls` to hide these “uninteresting” files from directory listings unless the `-a` option is used.

We recommend that you include default startup files for each shell that is popular on your systems so that users continue to have a reasonable default environment even if they change shells. Table 8.2 lists a variety of common startup files.

### Table 8.2: Common startup files and their uses

<table>
<thead>
<tr>
<th>Target</th>
<th>Filename</th>
<th>Typical uses</th>
</tr>
</thead>
<tbody>
<tr>
<td>all shells</td>
<td>.login_conf</td>
<td>Sets user-specific login defaults (FreeBSD)</td>
</tr>
<tr>
<td>sh</td>
<td>.profile</td>
<td>Sets search path, terminal type, and environment</td>
</tr>
<tr>
<td>bash¹</td>
<td>.bashrc</td>
<td>Sets the terminal type (if needed)</td>
</tr>
<tr>
<td></td>
<td></td>
<td>Sets biff and mesg switches</td>
</tr>
<tr>
<td></td>
<td>.bash_profile</td>
<td>Sets up environment variables</td>
</tr>
<tr>
<td></td>
<td></td>
<td>Sets command aliases</td>
</tr>
<tr>
<td></td>
<td></td>
<td>Sets the search path</td>
</tr>
<tr>
<td></td>
<td></td>
<td>Sets the umask value to control permissions</td>
</tr>
<tr>
<td></td>
<td></td>
<td>Sets CDPATH for filename searches</td>
</tr>
<tr>
<td></td>
<td></td>
<td>Sets the PS1 (prompt) and HISTCONTROL variables</td>
</tr>
<tr>
<td>csh/tcsh</td>
<td>.login</td>
<td>Read by “login” instances of csh</td>
</tr>
<tr>
<td></td>
<td>.cshrc</td>
<td>Read by all instances of csh</td>
</tr>
<tr>
<td>vi/viminfo</td>
<td>.vimrc/.viminfo</td>
<td>Sets vi/vim editor options</td>
</tr>
<tr>
<td>emacs</td>
<td>.emacs</td>
<td>Sets emacs editor options and key bindings</td>
</tr>
<tr>
<td>git</td>
<td>.gitconfig</td>
<td>Sets user, editor, color, and alias options for Git</td>
</tr>
<tr>
<td>GNOME</td>
<td>.gconf</td>
<td>GNOME user configuration via gconf</td>
</tr>
<tr>
<td></td>
<td>.gconfpath</td>
<td>Path for additional user configuration via gconf</td>
</tr>
<tr>
<td>KDE</td>
<td>.kde/</td>
<td>Directory of configuration files</td>
</tr>
</tbody>
</table>

¹. bash also reads .profile or /etc/profile in emulation of sh. The .bash_profile file is read by login shells, and the .bashrc file is read by interactive, non-login shells.
Sample startup files are traditionally kept in `/etc/skel`. If you customize your systems’ startup file examples, `/usr/local/etc/skel` is a reasonable place to put the modified copies.

The entries in Table 8.2 for the GNOME and KDE window environments are really just the beginning. In particular, take a look at `gconf`, which is the tool that stores application preferences for GNOME programs in a manner analogous to the Windows registry.

Make sure that the default shell files you give to new users set a reasonable default value for `umask`; we suggest 077, 027, or 022, depending on the friendliness and size of your site. If you do not assign new users to individual groups, we recommend `umask` 077, which gives the owner full access but the group and the rest of the world no access.

See [this page](#) for details on `umask`.

Depending on the user’s shell, `/etc` may contain system-wide startup files that are processed before the user’s own startup files. For example, `bash` and `sh` read `/etc/profile` before processing `~/.profile` and `~/.bash_profile`. These files are a good place in which to put site-wide defaults, but bear in mind that users can override your settings in their own startup files. For details on other shells, see the man page for the shell in question.

By convention, Linux also keeps fragments of startup files in the `/etc/profile.d` directory. Although the directory name derives from `sh` conventions, `/etc/profile.d` can actually include fragments for several different shells. The specific shells being targeted are distinguished by filename suffixes (`*.sh`, `*.csh`, etc.). There’s no magic `profile.d` support built into the shells themselves; the fragments are simply executed by the default startup script in `/etc` (e.g., `/etc/profile` in the case of `sh` or `bash`).

Separating the default startup files into fragments facilitates modularity and allows software packages to include their own shell-level defaults. For example, the `colorls.*` fragments coach shells on how to properly color the output of `ls` so as to make it unreadable on dark backgrounds.
**Setting home directory permissions and ownerships**

Once you’ve created a user’s home directory and copied in a reasonable default environment, turn the directory over to the user and make sure that the permissions on it are appropriate. The command

```
$ sudo chown -R newuser:newgroup ~newuser
```

sets ownerships properly. Note that you cannot use

```
$ sudo chown newuser:newgroup ~newuser/.*
```

to `chown` the dot files because `newuser` would then own not only his or her own files but also the parent directory “..” as well. This is a common and dangerous mistake.
Configuring roles and administrative privileges

Role-based access control (RBAC) allows system privileges to be tailored for individual users and is available on many of our example systems. RBAC is not a traditional part of the UNIX or Linux access control model, but if your site uses it, role configuration must be a part of the process of adding users. RBAC is covered in detail starting on this page in the Access Control and Rootly Powers chapter.

See Chapter 31 for more information about SOX and GLBA

Legislation such as the Sarbanes-Oxley Act, the Health Insurance Portability and Accountability Act (HIPAA), and the Gramm-Leach-Bliley Act in the United States has complicated many aspects of system administration in the corporate arena, including user management. Roles might be your only viable option for fulfilling some of the SOX, HIPAA, and GLBA requirements.
Finishing up

To verify that a new account has been properly configured, first log out, then log in as the new user and execute the following commands:

```bash
$ pwd  # To verify the home directory
$ ls -la  # To check owner/group of startup files
```

You need to notify new users of their login names and initial passwords. Many sites send this information by email, but that’s generally not a secure choice. Better options are to do it in person, over the phone, or through a text message. (If you are adding 500 new freshmen to the campus’s CS-1 machines, punt the notification problem to the instructor!) If you must distribute account passwords by email, make sure the passwords expire in a couple of days if they are not used and changed.

See [this page](#) for more information about written user contracts.

If your site requires users to sign a written policy agreement or appropriate use policy, be sure this step has been completed before you release a new account. This check prevents oversights and strengthens the legal basis of any sanctions you might later need to impose. This is also the time to point users toward additional documentation on local customs.

Remind new users to change their passwords immediately. You can enforce this by setting the password to expire within a short time. Another option is to have a script check up on new users and be sure their encrypted passwords have changed. Because the same password can have many encrypted representations, this method verifies only that the user has reset the password, not that it has actually been changed to a *different* password.

In environments where you know users personally, it’s relatively easy to keep track of who’s using a system and why. But if you manage a large and dynamic user base, you need a more formal way to keep track of accounts. Maintaining a database of contact information and account statuses helps you figure out, once the act of creating the account has faded from memory, who people are and why they have an account.
8.7 Scripts for Adding Users: useradd, adduser, and newusers

Our example systems all come with a useradd or adduser script that implements the basic procedure outlined above. However, these scripts are configurable, and you will probably want to customize them to fit your environment. Unfortunately, each system has its own idea of what you should customize, where you should implement the customizations, and what the default behavior should be. Accordingly, we cover these details in vendor-specific sections.

Table 8.3 is a handy summary of commands and configuration files related to managing users.

**Table 8.3: Commands and configuration files for user management**

<table>
<thead>
<tr>
<th>System</th>
<th>Commands</th>
<th>Configuration files</th>
</tr>
</thead>
<tbody>
<tr>
<td>All Linux</td>
<td>useradd, usermod, userdel</td>
<td>/etc/login.defs, /etc/default/useradd</td>
</tr>
<tr>
<td>Debian/Ubuntu*</td>
<td>adduser, deluser</td>
<td>/etc/adduser.conf, /etc/deluser.conf</td>
</tr>
<tr>
<td>FreeBSD</td>
<td>adduser, rmuser</td>
<td>/etc/login.conf</td>
</tr>
</tbody>
</table>

* a. This suite wraps the standard Linux version and includes a few more features.
**useradd on Linux**

Most Linux distributions include a basic `useradd` suite that draws its configuration parameters from both `/etc/login.defs` and `/etc/default/useradd`.

The `login.defs` file addresses issues such as password aging, choice of encryption algorithms, location of mail spool files, and the preferred ranges of UIDs and GIDs. You maintain the `login.defs` file by hand. The comments do a good job of explaining the various parameters.

Parameters stored in the `/etc/default/useradd` file include the location of home directories and the default shell for new users. You set these defaults through the `useradd` command itself. `useradd -D` prints the current values, and `-D` in combination with various other flags sets the values of specific options. For example,

```
$ sudo useradd -D -s /bin/bash
```

sets `bash` as the default shell.

Typical defaults are to put new users in individual groups, to use SHA-512 encryption for passwords, and to populate new users’ home directories with startup files from `/etc/skel`.

The basic form of the `useradd` command accepts the name of the new account on the command line:

```
$ sudo useradd hilbert
```

This command creates an entry similar to this one in `/etc/passwd`, along with a corresponding entry in the `shadow` file:

```
hilbert:x:1005:20::/home/hilbert:/bin/sh
```

`useradd` disables the new account by default. You must assign a real password to make the account usable.

A more realistic example is shown below. We specify that hilbert’s primary group should be “hilbert” and that he should also be added to the “faculty” group. We override the default home directory location and shell and ask `useradd` to create the home directory if it does not already exist:

```
$ sudo useradd -c "David Hilbert" -d /home/math/hilbert -g hilbert
-G faculty -m -s /bin/tcsh hilbert
```

This command creates the following `passwd` entry:

```
hilbert:x:1005:30:David Hilbert:/home/math/hilbert:/bin/tcsh
```
The assigned UID is one higher than the highest UID on the system, and the corresponding shadow entry is

hilbert:!14322:0:99999:7:0:0:

The password placeholder character(s) in the passwd and shadow file vary depending on the operating system. useradd also adds hilbert to the appropriate groups in /etc/group, creates the directory /home/math/hilbert with proper ownerships, and populates it from the /etc/skel directory.
adduser on Debian and Ubuntu

In addition to the useradd family of commands, the Debian lineage also supplies somewhat higher-level wrappers for these commands in the form of adduser and deluser. These add-on commands are configured in /etc/adduser.conf, where you can specify options such as these:

- Rules for locating home directories: by group, by username, etc.
- Permission settings for new home directories
- UID and GID ranges for system users and general users
- An option to create individual groups for each user
- Disk quotas (Boolean only, unfortunately)
- Regex-based matching of user names and group names

Other typical useradd parameters, such as rules for passwords, are set as parameters to the PAM module that does regular password authentication. (See this page for a discussion of PAM, aka Pluggable Authentication Modules.) adduser and deluser have twin cousins addgroup and delgroup.
adduser on FreeBSD

FreeBSD comes with adduser and rmuser shell scripts that you can either use as supplied or modify to fit your needs. The scripts are built on top of the facilities provided by the pw command.

adduser can be used interactively if you prefer. By default, it creates user and group entries and a home directory. You can point the script at a file containing a list of accounts to create with the -f flag, or enter in each user interactively.

For example, the process for creating a new user “raphael” looks like this:

```
$ sudo adduser
Username: raphael
Full name: Raphael Dobbins
Uid (Leave empty for default): <return>
Login group [raphael]: <return>
Login group is raphael. Invite raphael into other groups? [: <return>
Login class [default]: <return>
Shell (sh csh tcsh bash rbash nologin) [sh]: bash
Home directory [/home/raphael]: <return>
Home directory permissions (Leave empty for default): <return>
Use password–based authentication? [yes]: <return>
Use an empty password? (yes/no) [no]: <return>
Use a random password? (yes/no) [no]: yes
Lock out the account after creation? [no]: <return>
Username: raphael
Password: <random>
Full Name: Raphael Dobbins
Uid: 1004
Class: 
Groups: raphael
Home: /home/raphael
Home Mode: 
Shell: /usr/local/bin/bash
Locked: no
OK? (yes/no): yes
adduser: INFO: Successfully added (raphael) to the user database.
adduser: INFO: Password for (raphael) is: RSCAds5fyOwx0t
Add another user? (yes/no): no
Goodbye!
```
Linux’s newusers command creates multiple accounts at one time from the contents of a text file. It’s pretty gimpy, but it can be handy when you need to add a lot of users at once, such as when creating class-specific accounts. newusers expects an input file of lines just like the /etc/passwd file, except that the password field contains the initial password in clear text. Oops… better protect that file.

newusers honors the password aging parameters set in the /etc/login.defs file, but it does not copy in the default startup files as does useradd. The only startup file it copies in is .xauth.

At a university, what’s really needed is a batch adduser script that can use a list of students from enrollment or registration data to generate the input for newusers, with usernames formed according to local rules and guaranteed to be locally unique, with strong passwords randomly generated, and with UIDs and GIDs increasing for each user. You’re probably better off writing your own wrapper for useradd in Python than trying to get newusers to do what you need.
8.8 Safe Removal of a User’s Account and Files

When a user leaves your organization, that user’s login account and files must be removed from the system. If possible, don’t do that chore by hand; instead, let userdel or rmuser handle it. These tools ensure the removal of all references to the login name that were added by you or your useradd program. Once you’ve removed the remnants, use the following checklist to verify that all residual user data has been removed:

- Remove the user from any local user databases or phone lists.
- Remove the user from the mail aliases database, or add a forwarding address.
- Remove the user’s crontab file and any pending at jobs or print jobs.
- Kill any of the user’s processes that are still running.
- Remove the user from the passwd, shadow, group, and gshadow files.
- Remove the user’s home directory.
- Remove the user’s mail spool (if mail is stored locally).
- Clean up entries on shared calendars, room reservation systems, etc.
- Delete or transfer ownership of any mailing lists run by the deleted user.

Before you remove someone’s home directory, be sure to relocate any files that are needed by other users. You usually can’t be sure which files those might be, so it’s always a good idea to make an extra backup of the user’s home directory before deleting it.

Once you have removed all traces of a user, you may want to verify that the user’s old UID no longer owns files on the system. To find the paths of orphaned files, you can use the find command with the -nouser argument. Because find has a way of “escaping” onto network servers if you’re not careful, it’s usually best to check filesystems individually with -xdev:

```
$ sudo find filesystem -xdev -nouser
```

If your organization assigns individual workstations to users, it’s generally simplest and most efficient to re-image the entire system from a master template before turning the system over to a new user. Before you do the reinstallation, however, it’s a good idea to back up any local files on the system’s hard disk in case they are needed in the future. (Think license keys!)

Although all our example systems come with commands that automate the process of removing user presence, they probably do not do as thorough a job as you might like unless you have religiously extended them as you expanded the number of places in which user-related information is stored.
Debian and Ubuntu’s **deluser** is a Perl script that calls the usual **userdel**; it undoes all the things **adduser** does. It runs the script `/usr/local/sbin/deluser.local`, if it exists, to facilitate easy localization. The configuration file `/etc/deluser.conf` lets you set options such as these:

- Whether to remove the user’s home directory and mail spool
- Whether to back up the user’s files, and where to put the backup
- Whether to remove all files on the system owned by the user
- Whether to delete a group if it now has no members

Red Hat supports a **userdel.local** script but no pre- and post-execution scripts to automate sequence-sensitive operations such as backing up an about-to-be-removed user’s files.

FreeBSD’s **rmuser** script does a good job of removing instances of the user’s files and processes, a task that other vendors’ **userdel** programs do not even attempt.
8.9 User login lockout

On occasion, a user’s login must be temporarily disabled. A straightforward way to do this is to put a star or some other character in front of the user’s encrypted password in the `/etc/shadow` or `/etc/master.passwd` file. This measure prevents most types of password-regulated access because the password no longer decrypts to anything sensible.

FreeBSD lets you lock accounts with the `pw` command. A simple

```
$ sudo pw lock someuser
```

puts the string `*LOCKED*` at the start of the password hash, making the account unusable. Unlock the account by running

```
$ sudo pw unlock someuser
```

On all our Linux distributions, the `usermod -L user` and `usermod -U user` commands define an easy way to lock and unlock passwords. They are just shortcuts for the password twiddling described above: the `-L` puts an `!` in front of the encrypted password in the `/etc/shadow` file, and the `-U` removes it.

Unfortunately, modifying a user’s password simply makes logins fail. It does not notify the user of the account suspension or explain why the account no longer works. In addition, commands such as `ssh` that do not necessarily check the system password may continue to function.

An alternative way to disable logins is to replace the user’s shell with a program that prints an explanatory message and supplies instructions for rectifying the situation. The program then exits, terminating the login session.

This approach has both advantages and disadvantages. Any forms of access that check the password but do not pay attention to the shell will not be disabled. To facilitate the “disabled shell” trick, many daemons that afford nonlogin access to the system (e.g., `ftpd`) check to see if a user’s login shell is listed in `/etc/shells` and deny access if it is not. This is the behavior you want. Unfortunately, it’s not universal, so you may have to do some fairly comprehensive testing if you decide to use shell modification as a way of disabling accounts.

Another issue is that your carefully written explanation of the suspended account might never be seen if the user tries to log in through a window system or through a terminal emulator that does not leave output visible after a logout.
8.10 Risk reduction with PAM

Pluggable Authentication Modules (PAM) is covered in the *Single Sign-On* chapter starting here. PAM centralizes the management of the system’s authentication facilities through standard library routines. That way, programs like `login`, `sudo`, `passwd`, and `su` need not supply their own tricky authentication code. An organization can easily expand its authentication methods beyond passwords to options such as Kerberos, one-time passwords, ID dongles, or fingerprint readers. PAM reduces the risk inherent in writing secured software, allows administrators to set site-wide security policies, and defines an easy way to add new authentication methods to the system.

Adding and removing users doesn’t involve tweaking the PAM configuration, but the tools involved operate under PAM’s rules and constraints. In addition, many of the PAM configuration parameters are similar to those used by `useradd` or `usermod`. If you change a parameter as described in this chapter and `useradd` doesn’t seem to be paying attention to it, check to be sure the system’s PAM configuration isn’t overriding your new value.
8.11 Centralized Account Management

Some form of centralized account management is essential for medium-to-large enterprises of all types, be they corporate, academic, or governmental. Users need the convenience and security of a single login name, UID, and password across the site. Administrators need a centralized system that allows changes (such as account revocations) to be instantly propagated everywhere.

Such centralization can be achieved in a variety of ways, most of which (including Microsoft’s Active Directory system) involve LDAP, the Lightweight Directory Access Protocol, in some capacity. Options range from bare-bones LDAP installations based on open source software to elaborate commercial identity management systems that come with a hefty price tag.
**LDAP and Active Directory**

LDAP is a generalized, database-like repository that can store user management data as well as other types of data. It uses a hierarchical client/server model that supports multiple servers as well as multiple simultaneous clients. One of LDAP’s big advantages as a site-wide repository for login data is that it can enforce unique UIDs and GIDs across systems. It also plays well with Windows, although the reverse is only marginally true.

See the section starting on [this page](#) for more information about LDAP and LDAP implementations.

Microsoft’s Active Directory uses LDAP and Kerberos and can manage many kinds of data, including user information. It’s a bit egotistical and wants to be the boss if it is interacting with UNIX or Linux LDAP repositories. If you need a single authentication system for a site that includes Windows desktops as well as UNIX and Linux systems, it is probably easiest to let Active Directory be in control and to use your UNIX LDAP databases as secondary servers.

See [Chapter 17, Single Sign-On](#), for more information on integrating UNIX or Linux with LDAP, Kerberos, and Active Directory.
Application-level single sign-on systems

Application-level single sign-on systems balance user convenience with security. The idea is that a user can sign on once (to a login prompt, web page, or Windows box) and be authenticated at that time. The user then obtains authentication credentials (usually implicitly, so that no active management is required) which can be used to access other applications. The user only has to remember one login and password sequence instead of many.

This scheme allows credentials to be more complex since the user does not need to remember or even deal with them. That theoretically increases security. However, the impact of a compromised account is greater because one login gives an attacker access to multiple applications. These systems make your walking away from a desktop machine while still logged in a significant vulnerability. In addition, the authentication server becomes a critical bottleneck. If it’s down, all useful work grinds to a halt across the enterprise.

Although application-level SSO is a simple idea, it implies a lot of back-end complexity because the various applications and machines that a user might want to access must understand the authentication process and SSO credentials.

Several open source SSO systems exist:

- JOSSO, an open source SSO server written in Java
- CAS, the Central Authentication Service, from Yale (also Java)
- Shibboleth, an open source SSO distributed under the Apache 2 license

A host of commercial systems are also available, most of them integrated with identity management suites, which are covered in the next section.
Identity management systems

“Identity management” (sometimes referred to as IAM, for “identity and access management”) is a common buzzword in user management. In plain language, it means identifying the users of your systems, authenticating their identities, and granting privileges according to those authenticated identities. The standardization efforts in this realm are led by the World Wide Web Consortium and by The Open Group.

Commercial identity management systems combine several key UNIX concepts into a warm and fuzzy GUI replete with marketing jargon. Fundamental to all such systems is a database of user authentication and authorization data, often stored in LDAP format. Control is achieved with concepts such as UNIX groups, and limited administrative privileges are enforced through tools such as sudo. Most such systems have been designed with an eye toward regulations that mandate accountability, tracking, and audit trails.

There are many commercial systems in this space: Oracle’s Identity Management, Courion, Avatier Identity Management Suite (AIMS), VMware Identity Manager, and SailPoint’s IdentityIQ, to name a few. In evaluating identity management systems, look for capabilities in the following areas:

Oversight:

- Implement a secure web interface for management that’s accessible both inside and outside the enterprise.
- Support an interface through which hiring managers can request that accounts be provisioned according to role.
- Coordinate with a personnel database to automatically remove access for employees who are terminated or laid off.

Account management:

- Generate globally unique user IDs.
- Create, change, and delete user accounts across the enterprise, on all types of hardware and operating systems.
- Support a workflow engine; for example, tiered approvals before a user is given certain privileges.
- Make it easy to display all users who have a certain set of privileges. Ditto for the privileges granted to a particular user.
- Support role-based access control, including user account provisioning by role. Allow exceptions to role-based provisioning, including a workflow for the approval of
exceptions.

- Configure logging of all changes and administrative actions. Similarly, configure reports generated from logging data (by user, by day, etc.).

Ease of use:

- Let users change (and reset) their own passwords, with enforcement of rules for picking strong passwords.

- Enable users to change their passwords globally in one operation.

Consider also how the system is implemented at the point where authorizations and authentications actually take place. Does the system require a custom agent to be installed everywhere, or does it conform itself to the underlying systems?
Cloud computing is the practice of leasing computer resources from a pool of shared capacity. Users of cloud services provision resources on demand and pay a metered rate for whatever they consume. Businesses that embrace the cloud enjoy faster time to market, greater flexibility, and lower capital and operating expenses than businesses that run traditional data centers.

The cloud is the realization of “utility computing,” first conceived by the late computer scientist John McCarthy, who described the idea in a talk at MIT in 1961. Many technological advances since McCarthy’s prescient remarks have helped to bring the idea to fruition. To name just a few:

- Virtualization software reliably allocates CPU, memory, storage, and network resources on demand.
- Robust layers of security isolate users and virtual machines from each other, even as they share underlying hardware.
- Standardized hardware components enable the construction of data centers with vast power, storage, and cooling capacities.
- A reliable global network connects everything.

Cloud providers capitalize on these innovations and many others. They offer myriad services ranging from hosted private servers to fully managed applications. The leading cloud vendors are competitive, highly profitable, and growing rapidly.
This chapter introduces the motivations for moving to the cloud, fills in some background on a few major cloud providers, introduces some of the most important cloud services, and offers tips for controlling costs. As an even briefer introduction, the section *Clouds: VPS quick start by platform*, shows how to create cloud servers from the command line.

Several other chapters in this book include sections that relate to the management of cloud servers. *Table 9.1* lists some pointers.

**Table 9.1: Cloud topics covered elsewhere in this book**

<table>
<thead>
<tr>
<th>Heading</th>
</tr>
</thead>
<tbody>
<tr>
<td><em>Recovery of cloud systems</em> (bootstrapping-related issues for the cloud)*</td>
</tr>
<tr>
<td><em>Cloud networking</em> (TCP/IP networking for cloud platforms)*</td>
</tr>
<tr>
<td><em>Web hosting in the cloud</em></td>
</tr>
<tr>
<td><em>Packer</em> (using Packer to build OS images for the cloud)*</td>
</tr>
<tr>
<td><em>Container clustering and management</em> (especially the section on AWS ECS)*</td>
</tr>
<tr>
<td><em>CI/CD in practice</em> (a CI/CD pipeline example that uses cloud services)*</td>
</tr>
<tr>
<td><em>Commercial application monitoring tools</em> (monitoring tools for the cloud)*</td>
</tr>
</tbody>
</table>

In addition, *Chapter 23, Configuration Management*, is broadly applicable to the management of cloud systems.
9.1 The Cloud in Context

The transition from servers in private data centers to the now-ubiquitous cloud has been rapid and dramatic. Let’s take a look at the reasons for this stampede.

Cloud providers create technically advanced infrastructure that most businesses cannot hope to match. They locate their data centers in areas with inexpensive electric power and copious networking cross-connects. They design custom server chassis that maximize energy efficiency and minimize maintenance. They use purpose-built network infrastructure with custom hardware and software fine-tuned to their internal networks. They automate aggressively to allow rapid expansion and reduce the likelihood of human error.

Because of all this engineering effort (not to mention the normal economies of scale), the cost of running distributed computing services is much lower for a cloud provider than for a typical business with a small data center. Cost savings are reflected both in the price of cloud services and in the providers’ profits.

Layered on top of this hardware foundation are management features that simplify and facilitate the configuration of infrastructure. Cloud providers offer both APIs and user-facing tools that control the provisioning and releasing of resources. As a result, the entire life cycle of a system—or group of systems distributed on a virtual network—can be automated. This concept goes by the name “infrastructure as code,” and it contrasts starkly with the manual server procurement and provisioning processes of times past.

Elasticity is another major driver of cloud adoption. Because cloud systems can be programmatically requested and released, any business that has cyclic demand can optimize operating costs by adding more resources during periods of peak usage and removing extra capacity when it is no longer needed. The built-in autoscaling features available on some cloud platforms streamline this process.

Cloud providers have a global presence. With some planning and engineering effort, businesses can reach new markets by releasing services in multiple geographic areas. In addition, disaster recovery is easier to implement in the cloud because redundant systems can be run in separate physical locations.

See this page for more information about DevOps.

All these characteristics pair well with the DevOps approach to system administration, which emphasizes agility and repeatability. In the cloud, you’re no longer restricted by slow procurement or provisioning processes, and nearly everything can be automated.

Still, a certain mental leap is required when you don’t control your own hardware. One industry metaphor captures the sentiment neatly: servers should be treated as cattle, not as pets. A pet is
named, loved, and cared for. When the pet is sick, it’s taken to a veterinarian and nursed back to health. Conversely, cattle are commodities that are herded, traded, and managed in large quantities. Sick cattle are shot.

A cloud server is just one member of a herd, and to treat it otherwise is to ignore a basic fact of cloud computing: cloud systems are ephemeral, and they can fail at any time. Plan for that failure and you’ll be more successful at running a resilient infrastructure.

Despite all its advantages, the cloud is not a panacea for quickly reducing costs or improving performance. Directly migrating an existing enterprise application from a data center to a cloud provider (a so-called “lift and shift”) is unlikely to be successful without careful planning. Operational processes for the cloud are different, and they entail training and testing. Furthermore, most enterprise software is designed for static environments, but individual systems in the cloud should be treated as short-lived and unreliable. A system is said to be cloud native if it is reliable even in the face of unanticipated events.
9.2 Cloud platform choices

Multiple factors influence a site’s choice of cloud provider. Cost, past experience, compatibility with existing technology, security, or compliance requirements, and internal politics are all likely to play a role. The selection process can also be swayed by reputation, provider size, features, and of course, marketing.

Fortunately, there are a lot of cloud providers out there. We’ve chosen to focus on just three of the major public cloud providers: Amazon Web Services (AWS), Google Cloud Platform (GCP), and DigitalOcean (DO). In this section we mention a few additional options for you to consider. Table 9.2 enumerates the major players in this space.

Table 9.2: The most widely used cloud platforms

<table>
<thead>
<tr>
<th>Provider</th>
<th>Notable qualities</th>
</tr>
</thead>
<tbody>
<tr>
<td>DigitalOcean</td>
<td>Simple and reliable. Lovable API. Good for development.</td>
</tr>
<tr>
<td></td>
<td>Comprehensive big-data services.</td>
</tr>
<tr>
<td>IBM Softlayer</td>
<td>More like hosting than cloud. Has a global private network.</td>
</tr>
<tr>
<td>Microsoft Azure</td>
<td>A distant second in size. Has a history of outages. Possibly worth</td>
</tr>
<tr>
<td></td>
<td>consideration for Microsoft shops.</td>
</tr>
<tr>
<td>OpenStack</td>
<td>Modular DIY open source platform for building private clouds. AWS-compatible</td>
</tr>
<tr>
<td></td>
<td>APIs.</td>
</tr>
<tr>
<td>Rackspace</td>
<td>Public and private clouds running OpenStack. Offers managed services for AWS and</td>
</tr>
<tr>
<td></td>
<td>Azure. Fanatical support.</td>
</tr>
<tr>
<td>VMware vCloud Air</td>
<td>Buzzword-laden service for public, private, and hybrid clouds. Uses VMware</td>
</tr>
<tr>
<td></td>
<td>technology. Probably doomed.</td>
</tr>
</tbody>
</table>
Public, private, and hybrid clouds

In a public cloud, the vendor controls all the physical hardware and affords access to systems over the Internet. This setup relieves users of the burden of installing and maintaining hardware, but at the expense of less control over the features and characteristics of the platform. AWS, GCP, and DO are all public cloud providers.

Private cloud platforms are similar, but are hosted within an organization’s own data center or managed by a vendor on behalf of a single customer. Servers in a private cloud are single-tenant, not shared with other customers as in a public cloud.

Private clouds offer flexibility and programmatic control, just as public clouds do. They appeal to organizations that already have significant capital invested in hardware and engineers, especially those that value full control of their environment.

OpenStack is the leading open source system used to create private clouds. It receives financial and engineering support from enterprises such as AT&T, IBM, and Intel. Rackspace itself is one of the largest contributors to OpenStack.

A combination of public and private clouds is called a hybrid cloud. Hybrids can be useful when an enterprise is first migrating from local servers to a public cloud, for adding temporary capacity to handle peak loads, and for a variety of other organization-specific scenarios. Administrators beware: operating two distinct cloud presences in tandem increases complexity more than proportionally.

VMware’s vSphere Air cloud, based on vSphere virtualization technology, is a seamless hybrid cloud for customers that already use VMware virtualization in their on-premises data center. Those users can move applications to and from vCloud Air infrastructure quite transparently.

The term “public cloud” is a bit unfortunate, connoting as it does the security and hygiene standards of a public toilet. In fact, customers of public clouds are isolated from each other by multiple layers of hardware and software virtualization. A private cloud offers little or no practical security benefit over a public cloud.

In addition, operating a private cloud is an intricate and expensive prospect that should not be undertaken lightly. Only the largest and most committed organizations have the engineering resources and wallet needed to implement a robust, secure private cloud. And once implemented, a private cloud’s features usually fall short of those offered by commercial public clouds.

For most organizations, we recommend the public cloud over the private or hybrid options. Public clouds offer the highest value and easiest administration. For the remainder of this book, our cloud coverage is limited to public options. The next few sections present a quick overview of each of our example platforms.
Amazon Web Services

AWS offers scores of services, ranging from virtual servers (EC2) to managed databases and data warehouses (RDS and Redshift) to serverless functions that execute in response to events (Lambda). AWS releases hundreds of updates and new features each year. It has the largest and most active community of users. AWS is by far the largest cloud computing business.

From the standpoint of most users, AWS has essentially unlimited capacity. However, new accounts come with limits that control how much compute power you can requisition. These restrictions protect both Amazon and you, since costs can quickly spiral out of control if services aren’t properly managed. To increase your limits, you complete a form on the AWS support site. The service limit documentation itemizes the constraints associated with each service.

The on-line AWS documentation located at aws.amazon.com/documentation is authoritative, comprehensive, and well organized. It should be the first place you look when researching a particular service. The white papers that discuss security, migration paths, and architecture are invaluable for those interested in constructing robust cloud environments.
Google Cloud Platform

If AWS is the reigning champion of the cloud, Google is the would-be usurper. It competes for customers through nefarious tricks such as lowering prices and directly addressing customers’ AWS pain points.

The demand for engineers is so fierce that Google has been known to poach employees from AWS. In the past, they’ve hosted parties in conjunction with the AWS re:Invent conference in Las Vegas in an attempt to lure both talent and users. As the cloud wars unfold, customers ultimately benefit from this competition in the form of lower costs and improved features.

Google runs the most advanced global network in the world, a strength that benefits its cloud platform. Google data centers are technological marvels that feature many innovations to improve energy efficiency and reduce operational costs. Google is relatively transparent about its operations, and their open source contributions help advance the cloud industry. See google.com/about/datacenters for photos and facts about how Google’s data centers operate.

Despite its technical savvy, in some ways Google is a follower in the public cloud, not a leader. Google had released other cloud products as early as 2008, including App Engine, the first platform-as-a-service product. But Google’s strategy and the GCP brand were not well developed until 2012. At that point, GCP was already somewhat late to the game.

GCP’s services have many of the same features (and often the same names) as their AWS equivalents. If you’re familiar with AWS, you’ll find the GCP web interface to be somewhat different on the surface. However, the functionality underneath is strikingly similar.

We anticipate that GCP will gain market share in the years to come as it improves its products and builds customer trust. It has hired some of the brightest minds in the industry, and they’re bound to develop some innovative technologies. As consumers, we all stand to benefit.
DigitalOcean

DigitalOcean is a different breed of public cloud. Whereas AWS and GCP compete to serve the large enterprises and growth-focused startups, DigitalOcean courts small customers with simpler needs. Minimalism is the name of the game. We like DigitalOcean for experiments and proof-of-concept projects.

DigitalOcean offers data centers in North America, Europe, and Asia. There are several centers in each of these regions, but they are not directly connected and so cannot be considered availability zones (see this page). As a result, it’s considerably more difficult to build global, highly available production services on DigitalOcean than on AWS or Google.

DigitalOcean servers are called droplets. They are simple to provision from the command line or web console, and they boot quickly. DigitalOcean supplies images for all our example operating systems except Red Hat. It also has a handful of images for popular open source applications such as Cassandra, Drupal, Django, and GitLab.

DigitalOcean also has load balancer and block storage services. In Chapter 26, Continuous Integration and Delivery, we include an example of provisioning a DigitalOcean load balancer with two droplets using HashiCorp’s Terraform infrastructure provisioning tool.
9.3 Cloud Service Fundamentals

Cloud services are loosely grouped into three categories:

- **Infrastructure-as-a-Service (IaaS)**, in which users request raw compute, memory, network, and storage resources. These are typically delivered in the form of virtual private servers, aka VPSs. Under IaaS, users are responsible for managing everything above the hardware: operating systems, networking, storage systems, and their own software.

- **Platform-as-a-Service (PaaS)**, in which developers submit their custom applications packaged in a format specified by the vendor. The vendor then runs the code on the user’s behalf. In this model, users are responsible for their own code, while the vendor manages the OS and network.

- **Software-as-a-Service (SaaS)**, the broadest category, in which the vendor hosts and manages software and users pay some form of subscription fee for access. Users maintain neither the operating system nor the application. Almost any hosted web application (think WordPress) falls into the SaaS category.

Table 9.3 shows how each of these abstract models breaks down in terms of the layers involved in a complete deployment.

<table>
<thead>
<tr>
<th>Layer</th>
<th>Local(^a)</th>
<th>IaaS</th>
<th>PaaS</th>
<th>SaaS</th>
</tr>
</thead>
<tbody>
<tr>
<td>Application</td>
<td>✔️</td>
<td>✔️</td>
<td>✔️</td>
<td>✔️</td>
</tr>
<tr>
<td>Databases</td>
<td>✔️</td>
<td>✔️</td>
<td>✔️</td>
<td>✔️</td>
</tr>
<tr>
<td>Application runtime</td>
<td>✔️</td>
<td>✔️</td>
<td>✔️</td>
<td>✔️</td>
</tr>
<tr>
<td>Operating system</td>
<td>✔️</td>
<td>✔️</td>
<td></td>
<td></td>
</tr>
<tr>
<td>Virtual network, storage, and servers</td>
<td>✔️</td>
<td></td>
<td>✔️</td>
<td></td>
</tr>
<tr>
<td>Virtualization platform</td>
<td>✔️</td>
<td></td>
<td></td>
<td></td>
</tr>
<tr>
<td>Physical servers</td>
<td>✔️</td>
<td></td>
<td></td>
<td></td>
</tr>
<tr>
<td>Storage systems</td>
<td>✔️</td>
<td></td>
<td></td>
<td></td>
</tr>
<tr>
<td>Physical network</td>
<td>✔️</td>
<td></td>
<td></td>
<td></td>
</tr>
<tr>
<td>Power, space, and cooling</td>
<td>✔️</td>
<td></td>
<td></td>
<td></td>
</tr>
</tbody>
</table>

a. Local: local servers and network
   IaaS: Infrastructure-as-a-Service (virtual servers)
   PaaS: Platform-as-a-Service (e.g., Google App Engine)
   SaaS: Software-as-a-Service (e.g., most web-based services)

Of these options, IaaS is the most pertinent to system administration. In addition to defining virtual computers, IaaS providers virtualize the hardware elements that are typically connected to
them, such as disks (now described more generally as “block storage devices”) and networks. Virtual servers can inhabit virtual networks for which you specify the topology, routes, addressing, and other characteristics. In most cases, these networks are private to your organization.

IaaS can also include other core services such as databases, queues, key/value stores, and compute clusters. These features combine to create a complete replacement for (and in many cases, an improvement over) the traditional data center.

PaaS is an area of great promise that is not yet fully realized. Current offerings such as AWS Elastic Beanstalk, Google App Engine, and Heroku come with environmental constraints or nuances that render them impractical (or incomplete) for use in busy production environments. Time and again we’ve seen business outgrow these services. However, new services in this area are receiving a lot of attention. We anticipate dramatic improvements in the coming years.

Cloud providers differ widely in terms of their exact features and implementation details, but conceptually, many services are quite similar. The following sections describe cloud services generally, but because AWS is the front-runner in this space, we sometimes adopt its nomenclature and conventions as defaults.
Access to the cloud

Most cloud providers’ primary interface is some kind of web-based GUI. New system administrators should use this web console to create an account and to configure their first few resources.

Cloud providers also define APIs that access the same underlying functionality as that of the web console. In most cases, they also have a standard command-line wrapper, portable to most systems, for those APIs.

Even veteran administrators make frequent use of web GUIs. However, it’s also important to get friendly with the command-line tools because they lend themselves more readily to automation and repeatability. Use scripts to avoid the tedious and sluggish process of requesting everything through a browser.

Cloud vendors also maintain software development kits (SDKs) for many popular languages to help developers use their APIs. Third party tools use the SDKs to simplify or automate specific sets of tasks. You’ll no doubt encounter these SDKs if you write your own tools.

You normally use SSH with public key authentication to access UNIX and Linux systems running in the cloud. See SSH, the Secure SHell for more information about the effective use of SSH.

Some cloud providers let you access a console session through a web browser, which can be especially helpful if you mistakenly lock yourself out with a firewall rule or broken SSH configuration. It’s not a representation of the system’s actual console, though, so you can’t use this feature to debug bootstrapping or BIOS issues.
Regions and availability zones

Cloud providers maintain data centers around the world. A few standard terms describe geography-related features.

A “region” is a location in which a cloud provider maintains data centers. In most cases, regions are named after the territory of intended service even though the data centers themselves are more concentrated. For example, Amazon’s us-east-1 region is served by data centers in north Virginia.

It takes about 5ms for a fiber optic signal to travel 1,000km, so regions the size of the U.S. east coast are fine from a performance standpoint. The network connectivity available to a data center is more important than its exact location.

Some providers also have “availability zones” (or simply “zones”) which are collections of data centers within a region. Zones within a region are peered through high-bandwidth, low-latency, redundant circuits, so inter-zone communication is fast, though not necessarily cheap. Anecdotally, we’ve experienced inter-zone latency of less than 1ms.

Zones are typically designed to be independent of one another in terms of power and cooling, and they’re geographically dispersed so that a natural disaster that affects one zone has a low probability of affecting others in the same region.

Regions and zones are fundamental to building highly available network services. Depending on availability requirements, you can deploy in multiple zones and regions to minimize the impact of a failure within a data center or geographic area. Availability zone outages can occur, but are rare; regional outages are rarer still. Most services from cloud vendors are aware of zones and use them to achieve built-in redundancy.

Exhibit A: Servers distributed among multiple regions and zones
Multiregion deployments are more complex because of the physical distances between regions and the associated higher latency. Some cloud vendors have faster and more reliable inter-region networks than others. If your site serves users around the world, the quality of your cloud vendor’s network is paramount.

Choose regions according to geographic proximity to your user base. For scenarios in which the developers and users are in different geographic regions, consider running your development systems close to the developers and production systems closer to the users.

For sites that deliver services to a global user base, running in multiple regions can substantially improve performance for end users. Requests can be routed to each client’s regional servers by exploitation of geographic DNS resolution, which determines clients’ locations by their source IP addresses.

Most cloud platforms have regions for North America, South America, Europe, and the Asia Pacific countries. Only AWS and Azure have a direct presence in China. Some platforms, notably AWS and vCloud, have regions compatible with strict U.S. federal ITAR requirements.
Virtual private servers

The flagship service of the cloud is the virtual private server, a virtual machine that runs on the provider’s hardware. Virtual private servers are sometimes called instances. You can create as many instances as you need, running your preferred operating system and applications, then shut the instances down when they’re no longer needed. You pay only for what you use, and there’s typically no up-front cost.

Because instances are virtual machines, their CPU power, memory, disk size, and network settings can be customized when the instance is created and even adjusted after the fact. Public cloud platforms define preset configurations called instance types. They range from single-CPU nodes with 512MiB of memory to large systems with many CPU cores and multiple TiB of memory. Some instance types are balanced for general use, and others are specialized for CPU-, memory-, disk-, or network-intensive applications. Instance configurations are one area in which cloud vendors compete vigorously to match market needs.

Instances are created from “images,” the saved state of an operating system that contains (at minimum) a root filesystem and a boot loader. An image might also include disk volumes for additional filesystems and other custom settings. You can easily create custom images with your own software and settings.

All our example operating systems are widely used, so cloud platforms typically supply official images for them. (Currently, you must build your own FreeBSD image if you use Google Compute Engine.) Many third party software vendors also maintain cloud images that have their software preinstalled to facilitate adoption by customers. It’s also easy to create your own custom images. Learn more about how to create virtual machine images in Packer.
Networking

Cloud providers let you create virtual networks with custom topologies that isolate your systems from each other and from the Internet. On platforms that offer this feature, you can set the address ranges of your networks, define subnets, configure routes, set firewall rules, and construct VPNs to connect to external networks. Expect some network-related operational overhead and maintenance when building larger, more complex cloud deployments.

See this page for more information about VPNs.

See this page for more information about RFC1918 private addresses.

You can make your servers accessible to the Internet by leasing publicly routable addresses from your provider—all providers have a large pool of such addresses from which users can draw. Alternatively, servers can be given only a private RFC1918 address within the address space you selected for your network, rendering them publicly inaccessible.

Systems without public addresses are not directly accessible from the Internet, even for administrative attention. You can access such hosts through a jump server or bastion host that is open to the Internet, or through a VPN that connects to your cloud network. For security, the smaller the external-facing footprint of your virtual empire, the better.

Although this all sounds promising, you have even less control over virtual networks than you do over traditional networks, and you’re subject to the whims and vagaries of the feature set made available by your chosen provider. It’s particularly maddening when new features launch but can’t interact with your private network. (We’re looking at you, Amazon!)

Skip to this page for the details on TCP/IP networking in the cloud.
Storage

Data storage is a major part of cloud computing. Cloud providers have the largest and most advanced storage systems on the planet, so you’ll be hard pressed to match their capacity and capabilities in a private data center. The cloud vendors bill by the amount of data you store. They are highly motivated to give you as many ways as possible to ingest your data. (Case in point: AWS offers on-site visits from the AWS Snowmobile, a 45-foot long shipping container towed by a semi truck that can transfer 100 PiB from your data center to the cloud.)

Here are a few of the most important ways to store data in the cloud:

- **Object stores** contain collections of discrete objects (files, essentially) in a flat namespace. Object stores can accommodate a virtually unlimited amount of data with exceptionally high reliability but relatively slow performance. They are designed for a read-mostly access pattern. Files in an object store are accessed over the network through HTTPS. Examples include AWS S3 and Google Cloud Storage.

- Block storage devices are virtualized hard disks. They can be requisitioned at your choice of capacities and attached to a virtual server, much like SAN volumes on a traditional network. You can move volumes among nodes and customize their I/O profiles. Examples include AWS EBS and Google persistent disks.

- Ephemeral storage is local disk space on a VPS that is created from disk drives on the host server. These are normally fast and capacious, but the data is lost when you delete the VPS. Therefore, ephemeral storage is best used for temporary files. Examples include instance store volumes on AWS and local SSDs on GCP.

In addition to these raw storage services, cloud providers usually offer a variety of freestanding database services that you can access over the network. Relational databases such as MySQL, PostgreSQL, and Oracle run as services on the AWS Relational Database Service. They offer built-in multizone redundancy and encryption for data at rest.

Distributed analytics databases such as AWS Redshift and GCP BigQuery offer incredible ROI; both are worth a second look before you build your own expensive data warehouse. Cloud vendors also offer the usual assortment of in-memory and NoSQL databases such as Redis and memcached.
**Identity and authorization**

Administrators, developers, and other technical staff all need to manage cloud services. Ideally, access controls should conform to the principle of least privilege: each principal can access only the entities that are relevant to it, and nothing more. Depending on the context, such access control specifications can become quite elaborate.

AWS is exceptionally strong in this area. Their service, called Identity and Access Management (IAM), defines not only users and groups but also roles for systems. A server can be assigned policies, for example, to allow its software to start and stop other servers, store and retrieve data in an object store, or interact with queues—all with automatic key rotation. IAM also has an API for key management to help you store secrets safely.

Other cloud platforms have fewer authorization features. Unsurprisingly, Azure’s service is based on Microsoft’s Active Directory. It pairs well with sites that have an existing directory to integrate with. Google’s access control service, also called IAM, is relatively coarse-grained and incomplete in comparison with Amazon’s.
Automation

The APIs and CLI tools created by cloud vendors are the basic building blocks of custom automation, but they’re often clumsy and impractical for orchestrating larger collections of resources. For example, what if you need to create a new network, launch several VPS instances, provision a database, configure a firewall, and finally, connect all these components? Written in terms of a raw cloud API, that would make for a complex script.

AWS CloudFormation was the first service to address this problem. It accepts a template in JSON or YAML format that describes the desired resources and their associated configuration details. You submit the template to CloudFormation, which checks it for errors, sorts out dependencies among resources, and creates or updates the cloud configuration according to your specifications.

CloudFormation templates are powerful but error prone in human hands because of their strict syntax requirements. A complete template is unbearably verbose and a challenge for humans to even read. Instead of writing these templates by hand, we prefer to automatically render them with a Python library called Troposphere from Mark Peek (see github.com/cloudtools/troposphere).

Third party services also target this problem. Terraform, from the open source company HashiCorp, is a cloud-agnostic tool for constructing and changing infrastructure. As with CloudFormation, you describe resources in a custom template and then let Terraform make the proper API calls to implement your configuration. You can then check your configuration file into version control and manage the infrastructure over time.
Serverless functions

One of the most innovative features in the cloud since its emergence are the cloud function services, sometimes called functions-as-a-service, also referred to as “serverless” features. Cloud functions are a model of code execution that do not require any long-lived infrastructure. Functions execute in response to an event, such as the arrival of a new HTTP request or an object being uploaded to a storage location.

For example, consider a traditional web server. HTTP requests are forwarded by the networking stack of the operating system to a web server, which routes them appropriately. When the response completes, the web server continues to wait for requests.

Contrast this with the serverless model. An HTTP request arrives, and it triggers the cloud function to handle the response. When complete, the cloud function terminates. The owner pays for the period of time that the function executes. There is no server to maintain and no operating system to manage.

AWS introduced Lambda, their cloud function service, at a conference in 2014. Google followed shortly with a Cloud Functions service. Several cloud function implementations exist for projects like OpenStack, Mesos, and Kubernetes.

Serverless functions hold great promise for the industry. A massive ecosystem of tools is emerging to support simpler and more powerful use of the cloud. We’ve found many uses for these short-lived, serverless functions in our day-to-day administrative duties. We anticipate rapid advances in this area in the coming years.
The cloud is an excellent sandbox in which to learn UNIX and Linux. This short section helps you get up and running with virtual servers on AWS, GCP, or DigitalOcean. As system administrators, we rely extensively on the command line (as opposed to web GUIs) for interacting with the cloud, so we illustrate the use of those tools here.
**Amazon Web Services**

To use AWS, first set up an account at aws.amazon.com. Once you create the account, immediately follow the guidance in the AWS Trusted Advisor to configure your account according to the suggested best practices. You can then navigate to the individual service consoles for EC2, VPC, etc.

Each AWS service has a dedicated user interface. When you log in to the web console, you’ll see the list of services at the top. Within Amazon, each service is managed by an independent team, and the UI unfortunately reflects this fact. Although this decoupling has helped AWS services grow, it does lead to a somewhat fragmented user experience. Some interfaces are more refined and intuitive than others.

To protect your account, enable multifactor authentication (MFA) for the root user, then create a privileged IAM user for day-to-day use. We also generally configure an alias so that users can access the web console without entering an account number. This option is found on the landing page for IAM.

In the next section we introduce the official **aws** CLI tool written in Python. New users might also benefit from Amazon’s Lightsail quick start service, which aims to start an EC2 instance with minimum fuss.

**aws**: control AWS subsystems

---

See this page for more information about **pip**.

---

**aws** is a unified command-line interface to AWS services. It manages instances, provisions storage, edits DNS records, and performs most of the other tasks shown in the web console. The tool relies on the exceptional Boto library, a Python SDK for the AWS API, and it runs on any system with a working Python interpreter. Install it with **pip**:

```
$ pip install awscli
```

To use **aws**, first authenticate it to the AWS API by using a pair of random strings called the “access key ID” and the “secret access key.” You generate these credentials in the IAM web console and then copy-and-paste them locally.

Running **aws configure** prompts you to set your API credentials and default region:

```
$ aws configure
AWS Access Key ID: AKIAIOSFODNN7EXAMPLE
AWS Secret Access Key: wJalrXUtznFEMI/K7MDENG/bPxrFicyExampleKey
Default region name [us-east-1]: <return>
Default output format [None]: <return>
```
These settings are saved to ~/.aws/config. As long as you’re setting up your environment, we also recommend that you configure the bash shell’s autocompletion feature so that subcommands are easier to discover. See the AWS CLI docs for more information.

The first argument to aws names the specific service you want to manipulate; for example, ec2 for actions that control the Elastic Compute Cloud. You can add the keyword help at the end of any command to see instructions. For example, aws help, aws ec2 help, and aws ec2 describe-instances help all produce useful man pages.

Creating an EC2 instance

Use aws ec2 run-instances to create and launch EC2 instances. Although you can create multiple instances with a single command (by using the --count option), the instances must all share the same configuration. Here’s a minimal example of a complete command:

```
$ aws ec2 run-instances --image-id ami-d440a6e7
   --instance-type t2.nano --associate-public-ip-address
   --key-name admin-key
# output shown below
```

This example specifies the following configuration details:

- The base system image is an Amazon-supplied version of CentOS 7 named ami-d440a6e7. (AWS calls their images AMIs, for Amazon Machine Images.) Like other AWS objects, the image names are unfortunately not mnemonic; you must look up IDs in the EC2 web console or on the command line (aws ec2 describe-images) to decode them.

- The instance type is t2.nano, which is currently the smallest instance type. It has one CPU core and 512MiB of RAM. Details about the available instance types can be found in the EC2 web console.

- A preconfigured key pair is also assigned to control SSH access. You can generate a key pair with the ssh-keygen command (see this page), then upload the public key to the AWS EC2 console.

The output of that aws ec2 run-instances command is shown below. It’s JSON, so it’s easily consumed by other software. For example, after launching an instance, a script could extract the instance’s IP address and configure DNS, update an inventory system, or coordinate the launch of multiple servers.
By default, EC2 instances in VPC subnets do not have public IP addresses attached, rendering them accessible only from other systems within the same VPC. To reach instances directly from the Internet, use the `--associate-public-ip-address` option, as shown in our example command. You can discover the assigned IP address after the fact with `aws ec2 describe-instances` or by finding the instance in the web console.

Firewalls in EC2 are known as “security groups.” Because we didn’t specify a security group here, AWS assumes the “default” group, which allows no access. To connect to the instance, adjust the security group to permit SSH from your IP address. In real-world scenarios, security group structure should be carefully planned during network design. We discuss security groups in [Security groups and NACLs](#).

`aws configure` sets a default region, so you need not specify a region for the instance unless you want something other than the default. The AMI, key pair, and subnet are all region-specific, and `aws` complains if they don’t exist in the region you specify. (In this particular case, the AMI, key pair, and subnet are from the us-east-1 region.)

Take note of the `InstanceId` field in the output, which is a unique identifier for the new instance. You can use `aws ec2 describe-instances --instance-id id` to show details about an existing instance, or just use `aws ec2 describe-instances` to dump all instances in the default region.

Once the instance is running and the default security group has been adjusted to pass traffic on TCP port 22, you can use SSH to log in. Most AMIs are configured with a nonroot account that has `sudo` privileges. For Ubuntu the username is ubuntu; for CentOS, centos. FreeBSD and
Amazon Linux both use ec2-user. The documentation for your chosen AMI should specify the username if it’s not one of these.

See Chapter 8 for more information about user management.

Properly configured images allow only public keys for SSH authentication, not passwords. Once you’ve logged in with the SSH private key, you’ll have full `sudo` access with no password required. We recommend disabling the default user after the first boot and creating personal, named accounts.

**Viewing the console log**

Debugging low-level problems such as startup issues and disk errors can be challenging without access to the instance’s console. EC2 lets you retrieve the console output of an instance, which can be useful if the instance is in an error state or appears to be hung. You can do this through the web interface or with `aws ec2 get-console-output`, as shown:

```
$ aws ec2 get-console-output --instance-id i-c4f60303
{
  "InstanceId": "i-c4f60303",
  "Timestamp": "2015-12-21T00:01:45.000Z",
  "Output": "[ 0.000000] Initializing cgroup subsys cpuset\n  0.000000] Initializing cgroup subsys cpu\n[ 0.000000] Initializing cgroup subsys cpuacct\n[ 0.000000] Linux version 4.1.7-15.23.amzn1.x86_64 (mockbuild@gobi-build-60006) (gcc version 4.8.3 20140911 (Red Hat 4.8.3-9)) #1 SMP Mon Sep 14 23:20:33 UTC 2015\n...
}
```

The full log is of course much longer than this snippet. In the JSON dump, the contents of the log are unhelpfully concatenated as a single line. For better readability, clean it up with `sed`:

```
$ aws ec2 get-console-output --instance-id i-c4f60303 | sed 's/\r\n/\n/g'
{
  "InstanceId": "i-c4f60303",
  "Timestamp": "2015-12-21T00:01:45.000Z",
  "Output": "[ 0.000000] Initializing cgroup subsys cpuset
   [ 0.000000] Initializing cgroup subsys cpu
   [ 0.000000] Initializing cgroup subsys cpuacct
   [ 0.000000] Linux version 4.1.7-15.23.amzn1.x86_64
       (mockbuild@gobi-build-60006) (gcc version 4.8.3 20140911
        (Red Hat 4.8.3-9)) #1 SMP Mon Sep 14 23:20:33 UTC 2015
   ...
}
```
This log output comes directly from the Linux boot process. The example above shows a few lines from the moment the instance was first initialized. In most cases, you’ll find the most interesting information near the end of the log.

**Stopping and terminating instances**

When you’re finished with an instance, you can “stop” it to shut the instance down but retain it for later use, or “terminate” it to delete the instance entirely. By default, termination also releases the instance’s root disk into the ether. Once terminated, an instance can never be resurrected, even by AWS.

```
$ aws ec2 stop-instances --instance-id i-c4f60303
{
   "StoppingInstances": [
      
      "InstanceId": "i-c4f60303",
      "CurrentState": {
         "Code": 64,
         "Name": "stopping"
      },
      "PreviousState": {
         "Code": 16,
         "Name": "running"
      }
   ]
}
```

Note that virtual machines don’t change state instantly; it takes a minute for the hamsters to reset. Hence the presence of transitional states such as “starting” and “stopping.” Be sure to account for them in any instance-wrangling scripts you might write.
Google Cloud Platform

To get started with GCP, establish an account at cloud.google.com. If you already have a Google identity, you can sign up using the same account.

GCP services operate within a compartment known as a project. Each project has separate users, billing details, and API credentials, so you can achieve complete separation between disparate applications or areas of business. Once you create your account, create a project and enable individual GCP services according to your needs. Google Compute Engine, the VPS service, is one of the first services you might want to enable.

Setting up gcloud

`gcloud`, a Python application, is the CLI tool for GCP. It’s a component of the Google Cloud SDK, which contains a variety of libraries and tools for interfacing with GCP. To install it, follow the installation instructions at cloud.google.com/sdk.

Your first action should be to set up your environment by running `gcloud init`. This command starts a small, local web server and then opens a browser link to display the Google UI for authentication. After you authenticate yourself through the web browser, `gcloud` asks you (back in the shell) to select a project profile, a default zone, and other defaults. The settings are saved under `~/.config/gcloud/`.

Run `gcloud help` for general information or `gcloud -h` for a quick usage summary. Per-subcommand help is also available; for example, `gcloud help compute` shows a man page for the Compute Engine service.

Running an instance on GCE

Unlike `aws` commands, which return immediately, `gcloud compute` operates synchronously. When you run the `create` command to provision a new instance, for example, `gcloud` makes the necessary API call, then waits until the instance is actually up and running before it returns. This convention avoids the need to poll for the state of an instance after you create it. (See `aws ec2 wait` for information on polling for events or states within AWS EC2.)

To create an instance, first obtain the name or alias of the image you want to boot:

```
$ gcloud compute images list --regexp 'debian.*'
NAME                      PROJECT       ALIAS    DEPRECATED  STATUS
debian-7-wheezy-v20160119  debian-cloud  debian-7             READY
debian-8-jessie-v20160119  debian-cloud  debian-8             READY
```

Then create and boot the instance, specifying its name and the image you want:
The output normally has a column that shows whether the instance is “preemptible,” but in this case it was blank and we removed it to make the output fit on the page. Preemptible instances are less expensive than standard instances, but they can run for only 24 hours and can be terminated at any time if Google needs the resources for another purpose. They’re meant for long-lived operations that can tolerate interruptions, such as batch processing jobs.

Preemptible instances are similar in concept to EC2’s “spot instances” in that you pay a discounted rate for otherwise-spare capacity. However, we’ve found Google’s preemptible instances to be more sensible and simpler to manage than AWS’s spot instances. Long-lived standard instances remain the most appropriate choice for most tasks, however.

gcloud initializes the instance with a public and private IP address. You can use the public IP with SSH, but gcloud has a helpful wrapper to simplify SSH logins:

```
$ gcloud compute ssh ulsah
Last login: Mon Jan 25 03:33:48 2016
ulsah:~$
```

Cha-ching!
DigitalOcean

With advertised boot times of 55 seconds, DigitalOcean’s virtual servers (“droplets”) are the fastest route to a root shell. The entry level cost is $5 per month, so they won’t break the bank, either.

See this page for more details on setting up Ruby gems.

Once you create an account, you can manage your droplets through DigitalOcean’s web site. However, we find it more convenient to use tugboat, a command-line tool written in Ruby that uses DigitalOcean’s published API. Assuming that you have Ruby and its library manager, gem, installed on your local system, just run gem install tugboat to install tugboat.

See this page for more about SSH.

A couple of one-time setup steps are required. First, generate a pair of cryptographic keys that you can use to control access to your droplets:

$ ssh-keygen -t rsa -b 2048 -f ~/.ssh/id_rsa_do
Generating public/private rsa key pair.
Enter passphrase (empty for no passphrase): <return>
Enter same passphrase again: <return>
Your identification has been saved in /Users/ben/.ssh/id_rsa_do.
Your public key has been saved in /Users/ben/.ssh/id_rsa_do.pub.

Copy the contents of the public key file and paste them into DigitalOcean’s web console (currently under Settings → Security). As part of that process, assign a short name to the public key.

Next, connect tugboat to DigitalOcean’s API by entering an access token that you obtain from the web site. tugboat saves the token for future use in ~/.tugboat.

$ tugboat authorize
Note: You can get your Access Token from https://cloud.digitalocean.com/settings/tokens/new
Enter your access token: e9dff1a9a7ffdd8faf3...f37b015b3d459c2795b64
Enter your SSH key path (defaults to ~/.ssh/id_rsa): ~/.ssh/id_rsa_do
Enter your SSH user (defaults to root):
Enter your SSH port number (optional, defaults to 22):
Enter your default region (optional, defaults to nyc1): sfo1
...
Authentication with DigitalOcean was successful.
To create and start a droplet, first identify the name of the system image you want to use as a baseline. For example:

```
$ tugboat images | grep -i ubuntu
16.04.1 x64 (slug:, id: 21669205, distro: Ubuntu)
16.04.1 x64 (slug:, id: 22601368, distro: Ubuntu)
16.04.2 x64 (slug: ubuntu-16-04-x64, id: 23754420, distro: Ubuntu)
16.04.2 x32 (slug: ubuntu-16-04-x32, id: 23754441, distro: Ubuntu)
...
```

You also need DigitalOcean’s numeric ID for the SSH key you pasted into the web console:

```
$ tugboat keys
SSH Keys:
Name: id_rsa_do, (id: 1587367), fingerprint:
```

This output shows that the numeric ID for the key named id_rsa_do is 1587367. Create and start a droplet like this:

```
$ tugboat create -i ubuntu-16-04-x64 -k 1587367 ulsah-ubuntu
queueing creation of droplet 'ulsah-ubuntu'...Droplet created!
```

Here, the argument to `-k` is the SSH key ID, and the last argument is a short name for the droplet that you can assign as you wish.

Once the droplet has had time to boot, you can log in with `tugboat ssh`:

```
$ tugboat ssh ulsah-ubuntu
Droplet fuzzy name provided. Finding droplet ID...done, 23754420
  (ubuntu-16-04-x64)
Executing SSH on Droplet (ubuntu-16-04-x64)...
This droplet has a private IP, checking if you asked to use the
  Private IP...
You didn't! Using public IP for ssh...
Attempting SSH: root@45.55.1.165
Welcome to Ubuntu 16.04 ((GNU/Linux 4.4.0-28-generic x86_64)
root@ulsah-ubuntu:~#
```

You can create as many droplets as you need, but keep in mind that you’ll be billed for each one, even if it’s powered down. To inactivate a droplet, power it down, use `tugboat snapshot droplet-name snapshot-name` to memorialize the state of the system, and run `tugboat destroy droplet-name` to decommission the droplet. You can later recreate the droplet by using the snapshot as a source image.
9.5 Cost control

Cloud newcomers often naively anticipate that large-scale systems will be dramatically cheaper to run in the cloud than in a data center. This expectation might stem from the inverse sticker shock engendered by cloud platforms’ low, low price per instance-hour. Or perhaps the idea is implanted by the siren songs of cloud marketers, whose case studies always show massive savings.

Regardless of their source, it’s our duty to stamp out hope and optimism wherever they are found. In our experience, new cloud customers are often surprised when costs climb quickly.

Cloud tariffs generally consist of several components:

- The compute resources of virtual private servers, load balancers, and everything else that consumes CPU cycles to run your services. Pricing is per hour of use.
- Internet data transfer (both ingress and egress), as well as traffic among zones and regions. Pricing is per GiB or TiB transferred.
- Storage of all types: block storage volumes, object storage, disk snapshots, and in some cases, I/O to and from the various persistence stores. Pricing is per GiB or TiB stored per month.

For compute resources, the pay-as-you-go model, also known as “on-demand pricing,” is the most expensive. On AWS and DigitalOcean, the minimum billing increment is one hour, and on GCP it’s a minute. Prices range from fractions of a cent per hour (DigitalOcean’s smallest droplet type with 512MiB and one CPU core, or AWS t2.nano instances) to several dollars per hour (an i2.8xlarge instance on AWS with 32 cores, 104GiB RAM, and 8 x 800GB local SSDs).

You can realize substantial savings on virtual servers by paying up front for longer terms. On AWS, this is called “reserved instance pricing.” Unfortunately, it’s unbearably cumbersome and time-consuming to determine precisely what to purchase. Reserved EC2 instances are tied to a specific instance family. If you decide later that you need something different, your investment is lost. On the upside, if you reserve an instance, you are guaranteed that it will be available for your use. With on-demand instances, your desired type might not even be available when you go to provision it, depending on current capacity and demand. AWS continues to tweak its pricing structure, so with luck the current system might be simplified in the future.

For number crunching workloads that can tolerate interruptions, AWS offers spot pricing. The spot market is an auction. If your bid exceeds the current spot price, you’ll be granted use of the instance type you requested until the price exceeds your maximum bid, at which point your instance is terminated. The prices can be deeply discounted compared to the EC2 on-demand and reserved prices, but the use cases are limited.
Google Compute Engine pricing is refreshingly simple by comparison. Discounts are automatically applied for sustained use, and you never pay up front. You pay the full base price for the first week of the month, and the incremental price drops each week by 20% of the base rate, to a maximum discount of 60%. The net discount on a full month of use is 30%. That’s roughly comparable to the discount on a one-year reserved EC2 instance, but you can change instances at any time. (For the persnickety and the thrifty: because the discount scheme is linked to your billing cycle, the timing of transitions makes a difference. You can switch instance types at the start or end of a cycle with no penalty. The worst case is to switch halfway through a billing cycle, which incurs a penalty of about 20% of an instance’s monthly base rate.)

Network traffic can be even more difficult to predict reliably. The culprits commonly found to be responsible for high data-transfer costs include

- Web sites that ingest and serve large media files (videos, images, PDFs, and other large documents) directly from the cloud, rather than offloading them to a CDN (see this page)

- Inter-zone or inter-region traffic for database clusters that replicate for fault tolerance; for example, software such as Cassandra, MongoDB, and Riak

- MapReduce or data warehouse clusters that span multiple zones

- Disk images and volume snapshots transferred between zones or regions for backup (or by some other automated process)

In situations where replication among multiple zones is important for availability, you’ll save on transfer expenses by limiting clusters to two zones rather than using three or more. Some software offers tweaks such as compression that can reduce the amount of replicated data.

One substantial source of expense on AWS is provisioned IOPS for EBS volumes. Pricing for EBS is per GiB-month and IOPS-month. The price of a 200GiB EBS volume with 5,000 IOPS is a few hundred dollars per month. A cluster of these just might break the bank.

The best defense against high bills is to measure, monitor, and avoid overprovisioning. Use autoscaling features to remove capacity when it isn’t needed, lowering costs at times of low demand. Use more, smaller instances for more fine-grained control. Watch usage patterns carefully before spending a bundle on reserved instances or high-bandwidth volumes. The cloud is flexible, and you can make changes to your infrastructure as needed.

As environments grow, identifying where money is being spent can be a challenge. Larger cloud accounts might benefit from third party services that analyze use and offer tracking and reporting features. The two that we’ve used are Cloudability and CloudHealth. Both tap in to the billing features of AWS to break down reports by user-defined tag, service, or geographic location.
9.6 Recommended Reading


Barr, Jeff, and others at Amazon Web Services. aws.amazon.com/blogs/aws. The official blog of Amazon Web Services.


Bias, Randy. cloudscaling.com/blog. Randy Bias is a director at OpenStack and has insightful info on the private cloud industry and its future.

Cantrill, Bryan. *The Observation Deck*. dtrace.org/blogs/bmc. Interesting views and technical thoughts on general computing from the CTO of Joyent, a niche but interesting cloud platform.

Amazon. youtube.com/AmazonWebServices. Conference talks and other video content from AWS.
System daemons, the kernel, and custom applications all emit operational data that is logged and eventually ends up on your finite-sized disks. This data has a limited useful life and may need to be summarized, filtered, searched, analyzed, compressed, and archived before it is eventually discarded. Access and audit logs may need to be managed closely according to regulatory retention rules or site security policies.

A log message is usually a line of text with a few properties attached, including a time stamp, the type and severity of the event, and a process name and ID (PID). The message itself can range from an innocuous note about a new process starting up to a critical error condition or stack trace. It’s the responsibility of system administrators to glean useful, actionable information from this ongoing torrent of messages.

This task is known generically as log management, and it can be divided into a few major subtasks:

- Collecting logs from a variety of sources
- Providing a structured interface for querying, analyzing, filtering, and monitoring messages
Managing the retention and expiration of messages so that information is kept as long as it is potentially useful or legally required, but not indefinitely

UNIX has historically managed logs through an integrated but somewhat rudimentary system, known as syslog, that presents applications with a standardized interface for submitting log messages. Syslog sorts messages and saves them to files or forwards them to another host over the network. Unfortunately, syslog tackles only the first of the logging chores listed above (message collection), and its stock configuration differs widely among operating systems.

Perhaps because of syslog’s shortcomings, many applications, network daemons, startup scripts, and other logging vigilantes bypass syslog entirely and write to their own ad hoc log files. This lawlessness has resulted in a complement of logs that varies significantly among flavors of UNIX and even among Linux distributions.

Linux’s **systemd** journal represents a second attempt to bring sanity to the logging madness. The journal collects messages, stores them in an indexed and compressed binary format, and furnishes a command-line interface for viewing and filtering logs. The journal can stand alone, or it can coexist with the syslog daemon with varying degrees of integration, depending on the configuration.

A variety of third party tools (both proprietary and open source) address the more complex problem of curating messages that originate from a large network of systems. These tools feature such aids as graphical interfaces, query languages, data visualization, alerting, and automated anomaly detection. They can scale to handle message volumes on the order of terabytes per day. You can subscribe to these products as a cloud service or host them yourself on a private network.

**Exhibit A** depicts the architecture of a site that uses all the log management services mentioned above. Administrators and other interested parties can run a GUI against the centralized log cluster to review log messages from systems across the network. Administrators can also log in to individual nodes and access messages through the **systemd** journal or the plain text files written by syslog. If this diagram raises more questions than answers for you, you’re reading the right chapter.
When debugging problems and errors, experienced administrators turn to the logs sooner rather than later. Log files often contain important hints that point toward the source of vexing configuration errors, software bugs, and security issues. Logs are the first place you should look when a daemon crashes or refuses to start, or when a chronic error plagues a system that is trying to boot.

The importance of having a well-defined, site-wide logging strategy has grown along with the adoption of formal IT standards such as PCI DSS, COBIT, and ISO 27001, as well as with the maturing of regulations for individual industries. Today, these external standards may require you to maintain a centralized, hardened, enterprise-wide repository for log activity, with time stamps validated by NTP and with a strictly defined retention schedule. However, even sites without regulatory or compliance requirements can benefit from centralized logging.

This chapter covers the native log management software used on Linux and FreeBSD, including syslog, the systemd journal, and logrotate. We also introduce some additional tools for centralizing and analyzing logs across the network. The chapter closes with some general advice for setting up a sensible site-wide log management policy.
10.1 **Log locations**

UNIX is often criticized for being inconsistent, and indeed it is. Just take a look at a directory of log files and you’re sure to find some with names like *maillog*, some like *cron.log*, and some that use various distribution- and daemon-specific naming conventions. By default, most of these files are found in `/var/log`, but some renegade applications write their log files elsewhere on the filesystem.

**Table 10.1** compiles information about some of the more common log files on our example systems. The table lists the following:

- The log files to archive, summarize, or truncate
- The program that creates each
- An indication of how each filename is specified
- The frequency of cleanup that we consider reasonable
- The systems (among our examples) that use the log file
- A description of the file’s contents

Filenames in **Table 10.1** are relative to `/var/log` unless otherwise noted. Syslog maintains many of the listed files, but others are written directly by applications.

**Table 10.1: Log files on parade**
Log files are generally owned by root, although conventions for the ownership and mode of log files vary. In some cases, a less privileged process such as `httpd` may require write access to the log, in which case the ownership and mode should be set appropriately. You might need to use `sudo` to view log files that have tight permissions.

See [this page](#) for an introduction to disk partitioning.

Log files can grow quickly, especially the ones for busy services such as web, database, and DNS servers. An out-of-control log file can fill up the disk and bring the system to its knees. For this reason, it’s often helpful to define `/var/log` as a separate disk partition or filesystem. (Note that

<table>
<thead>
<tr>
<th>File</th>
<th>Program</th>
<th>Where</th>
<th>Frequency</th>
<th>Systems</th>
<th>Contents</th>
</tr>
</thead>
<tbody>
<tr>
<td>apache2/*</td>
<td>httpd</td>
<td>F</td>
<td>D</td>
<td>D</td>
<td>Apache HTTP server logs (v2)</td>
</tr>
<tr>
<td>apt*</td>
<td>APT</td>
<td>F</td>
<td>M</td>
<td>D</td>
<td>Aptitude package installations</td>
</tr>
<tr>
<td>auth.log</td>
<td>sudo, etc.</td>
<td>S</td>
<td>M</td>
<td>DF</td>
<td>Authorizations</td>
</tr>
<tr>
<td>boot.log</td>
<td>rc scripts</td>
<td>F</td>
<td>M</td>
<td>R</td>
<td>Output from system startup scripts</td>
</tr>
<tr>
<td>cloud-init.log</td>
<td>cloud-init</td>
<td>F</td>
<td>–</td>
<td>–</td>
<td>Output from cloud init scripts</td>
</tr>
<tr>
<td>cron, cron/log</td>
<td>cron</td>
<td>S</td>
<td>W</td>
<td>RF</td>
<td>cron executions and errors</td>
</tr>
<tr>
<td>daemon.log</td>
<td>various</td>
<td>S</td>
<td>W</td>
<td>D*</td>
<td>All daemon facility messages</td>
</tr>
<tr>
<td>debug*</td>
<td>various</td>
<td>S</td>
<td>D</td>
<td>F,D*</td>
<td>Debugging output</td>
</tr>
<tr>
<td>dmesg</td>
<td>kernel</td>
<td>H</td>
<td>–</td>
<td>all</td>
<td>Dump of kernel message buffer</td>
</tr>
<tr>
<td>dpkg.log</td>
<td>dpkg</td>
<td>F</td>
<td>M</td>
<td>D</td>
<td>Package management log</td>
</tr>
<tr>
<td>faillog*</td>
<td>login</td>
<td>H</td>
<td>W</td>
<td>D*</td>
<td>Failed login attempts</td>
</tr>
<tr>
<td>httpd/*</td>
<td>httpd</td>
<td>F</td>
<td>D</td>
<td>R</td>
<td>Apache HTTP server logs</td>
</tr>
<tr>
<td>kern.log</td>
<td>kernel</td>
<td>S</td>
<td>W</td>
<td>D</td>
<td>All kern facility messages</td>
</tr>
<tr>
<td>lastlog</td>
<td>login</td>
<td>H</td>
<td>–</td>
<td>R</td>
<td>Last login time per user (binary)</td>
</tr>
<tr>
<td>mail*</td>
<td>mail-related</td>
<td>S</td>
<td>W</td>
<td>RF</td>
<td>All mail facility messages</td>
</tr>
<tr>
<td>messages</td>
<td>various</td>
<td>S</td>
<td>W</td>
<td>R</td>
<td>The main system log file</td>
</tr>
<tr>
<td>samba/*</td>
<td>smbd, etc.</td>
<td>F</td>
<td>W</td>
<td>–</td>
<td>Samba (Windows/SMB file sharing)</td>
</tr>
<tr>
<td>secure</td>
<td>sshd, etc.</td>
<td>S</td>
<td>M</td>
<td>R</td>
<td>Private authorization messages</td>
</tr>
<tr>
<td>syslog*</td>
<td>various</td>
<td>S</td>
<td>W</td>
<td>D</td>
<td>The main system log file</td>
</tr>
<tr>
<td>wtmp</td>
<td>login</td>
<td>H</td>
<td>M</td>
<td>RD</td>
<td>Login records (binary)</td>
</tr>
<tr>
<td>xen/*</td>
<td>Xen</td>
<td>F</td>
<td>1m</td>
<td>RD</td>
<td>Xen virtual machine information</td>
</tr>
<tr>
<td>Xorg.*.log</td>
<td>Xorg</td>
<td>F</td>
<td>W</td>
<td>R</td>
<td>X Windows server errors</td>
</tr>
<tr>
<td>yum.log</td>
<td>yum</td>
<td>F</td>
<td>M</td>
<td>R</td>
<td>Package management log</td>
</tr>
</tbody>
</table>

a. Where: F = Configuration file, H = Hardwired, S = Syslog

b. Frequency: D = Daily, M = Monthly, NNm = Size-based (in MB, e.g., 1m), W = Weekly

c. Systems: D = Debian and Ubuntu (D* = Debian only), R = Red Hat and CentOS, F = FreeBSD

d. Binary file that must be read with the `faillog` utility
this advice is just as relevant to cloud-based instances and private virtual machines as it is to physical servers.)
Files not to manage

Most logs are text files to which lines are written as interesting events occur. But a few of the logs listed in Table 10.1 have a rather different context.

wtmp (sometimes wtmpx) contains a record of users’ logins and logouts as well as entries that record when the system was rebooted or shut down. It’s a fairly generic log file in that new entries are simply added to the end of the file. However, the wtmp file is maintained in a binary format. Use the last command to decode the information.

lastlog contains information similar to that in wtmp, but it records only the time of last login for each user. It is a sparse, binary file that’s indexed by UID. It will stay smaller if your UIDs are assigned in some kind of numeric sequence, although this is certainly nothing to lose sleep over in the real world. lastlog doesn’t need to be rotated because its size stays constant unless new users log in.

Finally, some applications (notably, databases) create binary transaction logs. Don’t attempt to manage these files. Don’t attempt to view them, either, or you’ll be treated to a broken terminal window.
How to view logs in the systemd journal

See the sections starting on this page for more information about systemd and systemd units.

For Linux distributions running systemd, the quickest and easiest way to view logs is to use the journalctl command, which prints messages from the systemd journal. You can view all messages in the journal, or pass the -u flag to view the logs for a specific service unit. You can also filter on other constraints such as time window, process ID, or even the path to a specific executable.

For example, the following output shows journal logs from the SSH daemon:

```
$ journalctl -u ssh
-- Logs begin at Sat 2016-08-27 23:18:17 UTC, end at Sat 2016-08-27 23:33:20 UTC. --
Aug 27 23:18:24 uxenial sshd[2230]: Server listening on 0.0.0.0 port 22.
Aug 27 23:18:24 uxenial sshd[2230]: Server listening on :: port 22.
Aug 27 23:18:24 uxenial systemd[1]: Starting Secure Shell server...
Aug 27 23:18:28 uxenial sshd[2326]: Accepted publickey for bwhaley from 10.0.2.2 port 60341 ssh2: RSA SHA256:aaRfGd10untn758+UCpxL7gkSwcszkaye/wukrdBATc
Aug 27 23:18:28 uxenial sshd[2326]: pam_unix(sshd:session): session opened for user bwhaley by (uid=0)
Aug 27 23:18:34 uxenial sshd[2480]: Did not receive identification string from 10.0.2.2
```

Use journalctl -f to print new messages as they arrive. This is the systemd equivalent of the much-beloved tail -f for following plain text files as they are being appended to.

The next section covers the systemd-journald daemon and its configuration.
10.2 The systemd Journal

In accordance with its mission to replace all other Linux subsystems, systemd includes a logging daemon called systemd-journald. It duplicates most of syslog’s functions but can also run peacefully in tandem with syslog, depending on how you or the system have configured it. If you’re leery of switching to systemd because syslog has always “just worked” for you, spend some time to get to know systemd. After a little practice, you may be pleasantly surprised.

Unlike syslog, which typically saves log messages to plain text files, the systemd journal stores messages in a binary format. All message attributes are indexed automatically, which makes the log easier and faster to search. As discussed above, you can use the journalctl command to review messages stored in the journal.

The journal collects and indexes messages from several sources:

- The /dev/log socket, to harvest messages from software that submits messages according to syslog conventions
- The device file /dev/kmsg, to collect messages from the Linux kernel. The systemd journal daemon replaces the traditional klogd process that previously listened on this channel and formerly forwarded the kernel messages to syslog.
- The UNIX socket /run/systemd/journal/stdout, to service software that writes log messages to standard output
- The UNIX socket /run/systemd/journal/socket, to service software that submits messages through the systemd journal API
- Audit messages from the kernel’s auditd daemon

Intrepid administrators can use the systemd-journal-remote utility (and its relatives, systemd-journal-gateway and systemd-journal-upload,) to stream serialized journal messages over the network to a remote journal. Unfortunately, this feature does not come preinstalled on vanilla distributions. As of this writing, packages are available for Debian and Ubuntu but not for Red Hat or CentOS. We expect this lapse to be rectified soon; in the meantime, we recommend sticking with syslog if you need to forward log messages among systems.
Configuring the systemd journal

The default journal configuration file is `/etc/systemd/journald.conf`; however, this file is not intended to be edited directly. Instead, add your customized configurations to the `/etc/systemd/journald.conf.d` directory. Any files placed there with a `.conf` extension are automatically incorporated into the configuration. To set your own options, create a new `.conf` file in this directory and include the options you want.

The default `journald.conf` includes a commented-out version of every possible option, along with each option’s default value, so you can see at a glance which options are available. They include the maximum size of journal, the retention period for messages, and various rate-limiting settings.

The Storage option controls whether to save the journal to disk. The possible values are somewhat confusing:

- volatile stores the journal in memory only.
- persistent saves the journal in `/var/log/journal/`, creating the directory if it doesn’t already exist.
- auto saves the journal in `/var/log/journal/` but does not create the directory. This is the default value.
- none discards all log data.

Most Linux distributions (including all our examples) default to the value auto and do not come with a `/var/log/journal` directory. Hence, the journal is not saved between reboots by default, which is unfortunate.

You can modify this behavior either by creating the `/var/log/journal` directory or by updating the journal to use persistent storage and restarting `systemd-journald`:

```
# mkdir /etc/systemd/journald.conf.d/
# cat << END > /etc/systemd/journald.conf.d/storage.conf
[Journal]
  Storage=persistent
END
# systemctl restart systemd-journald
```

This series of commands creates the custom configuration directory `journald.conf.d`, creates a configuration file to set the Storage option to persistent, and restarts the journal so that the new settings take effect. `systemd-journald` will now create the directory and retain the journal. We recommend this change for all systems; it’s a real handicap to lose all log data every time the system reboots.
One of the niftiest journal options is Seal, which enables Forward Secure Sealing (FSS) to increase the integrity of log messages. With FSS enabled, messages submitted to the journal cannot be altered without access to a cryptographic key pair. You generate the key pair itself by running `journalctl --setup-keys`. Refer to the man pages for `journald.conf` and `journalctl` for the full scoop on this option.
Adding more filtering options for journalctl

We showed a quick example of a basic journalctl log search here. In this section, we show some additional ways to use journalctl to filter messages and gather information about the journal.

To allow normal users to read from the journal without needing sudo permissions, add them to the systemd-journal UNIX group.

The --disk-usage option shows the size of the journal on disk:

```bash
# journalctl --disk-usage
Journals take up 4.0M on disk.
```

The --list-boots option shows a sequential list of system boots with numerical identifiers. The most recent boot is always 0. The dates at the end of the line show the time stamps of the first and last messages generated during that boot.

```bash
# journalctl --list-boots
-1 ce0... Sun 2016-11-13 18:54:42 UTC–Mon 2016-11-14 00:09:31
  0 844... Mon 2016-11-14 00:09:38 UTC–Mon 2016-11-14 00:12:56
```

You can use the -b option to restrict the log display to a particular boot session. For example, to view logs generated by SSH during the current session:

```bash
# journalctl -b 0 -u ssh
```

To show all the messages from yesterday at midnight until now:

```bash
# journalctl --since=yesterday --until=now
```

To show the most recent 100 journal entries from a specific binary:

```bash
# journalctl -n 100 /usr/sbin/sshd
```

You can use journalctl --help as a quick reference for these arguments.
Coexisting with syslog

Both syslog and the systemd journal are active by default on each of our example Linux systems. Both packages collect and store log messages. Why would you want both of them running, and how does that even work?

Unfortunately, the journal is missing many of the features that are available in syslog. As the discussion starting on this page demonstrates, rsyslog can receive messages from a variety of input plug-ins and forward them to a diverse set of outputs according to filters and rules, none of which is possible when the systemd journal is used. The systemd universe does include a remote streaming tool, systemd-journal-remote, but it’s relatively new and untested in comparison with syslog. Administrators may also find it convenient to keep certain log files in plain text, as syslog does, instead of in the journal’s binary format.

We anticipate that over time, new features in the journal will usurp syslog’s responsibilities. But for now, Linux distributions still need to run both systems to achieve full functionality.

The mechanics of the interaction between the systemd journal and syslog are somewhat convoluted. To begin with, systemd-journald takes over responsibility for collecting log messages from /dev/log, the logging socket that was historically controlled by syslog. (More specifically, the journal links /dev/log to /run/systemd/journal/dev-log.) For syslog to get in on the logging action, it must now access the message stream through systemd. Syslog can retrieve log messages from the journal in two ways:

- The systemd journal can forward messages to another socket (typically /run/systemd/journal/syslog), from which the syslog daemon can read them. In this mode of operation, systemd-journald simulates the original message submitters and conforms to the standard syslog API. Therefore, only the basic message parameters are forwarded; some systemd-specific metadata is lost.

- Alternatively, syslog can consume messages directly from the journal API, in the same manner as the journalctl command. This method requires explicit support for cooperation on the part of syslogd, but it’s a more complete form of integration that preserves the metadata for each message. (See man systemd.journal-fields for a rundown of the available metadata.)

Debian and Ubuntu default to the former method, but Red Hat and CentOS use the latter. To determine which type of integration has been configured on your system, inspect the ForwardToSyslog option in /etc/systemd/journald.conf. If its value is yes, socket-forwarding is in use.
10.3 Syslog

Syslog, originally written by Eric Allman, is a comprehensive logging system and IETF-standard logging protocol. RFC5424 is the latest version of the syslog specification, but the previous version, RFC3164, may better reflect the real-world installed base.

Syslog has two important functions: to liberate programmers from the tedious mechanics of writing log files, and to give administrators control of logging. Before syslog, every program was free to make up its own logging policy. System administrators had no consistent control over what information was kept or where it was stored.

Syslog is flexible. It lets administrators sort messages by source (“facility”) and importance (“severity level”) and route them to a variety of destinations: log files, users’ terminals, or even other machines. It can accept messages from a wide variety of sources, examine the attributes of the messages, and even modify their contents. Its ability to centralize the logging for a network is one of its most valuable features.

On Linux systems, the original syslog daemon (syslogd) has been replaced with a newer implementation called rsyslog (rsyslogd). Rsyslog is an open source project that extends the capabilities of the original syslog but maintains backward API compatibility. It is the most reasonable choice for administrators working on modern UNIX and Linux systems and is the only version of syslog we cover in this chapter.

⁻ Rsyslog is available for FreeBSD, and we recommend that you adopt it in preference to the standard FreeBSD syslog unless you have simple needs. For instructions on converting a FreeBSD system to use rsyslog, see wiki.rsyslog.com/index.php/FreeBSD. If you decide to stick with FreeBSD’s traditional syslog, jump to this page for configuration information.
Reading syslog messages

You can read plaintext messages from syslog with normal UNIX and Linux text processing tools such as `grep`, `less`, `cat`, and `awk`. The snippet below shows typical events in `/var/log/syslog` from a Debian host:

```
jessie# cat /var/log/syslog
Jul 16 19:43:01 jessie networking[244]: bound to 10.0.2.15 -- renewal in 42093 seconds.
Jul 16 19:43:01 jessie rpcbind[397]: Starting rpcbind daemon....
Jul 16 19:43:01 jessie nfs-common[412]: Starting NFS common utilities: statd idmapd.
Jul 16 19:43:01 jessie cron[436]: (CRON) INFO (pidfile fd = 3)
Jul 16 19:43:01 jessie cron[436]: (CRON) INFO (Running @reboot jobs)
Jul 16 19:43:01 jessie acpid: starting up with netlink and the input layer
Jul 16 19:43:01 jessie docker[486]: time="2016-07-16T19:43:01.972878480Z" level=info msg="Daemon has completed initialization"
Jul 16 19:43:01 jessie docker[486]: time="2016-07-16T19:43:01.972896608Z" level=info msg="Docker daemon commit=c3959b1 execdriver=native-0.2 graphdriver=aufs version=1.10.2"
Jul 16 19:43:01 jessie docker[486]: time="2016-07-16T19:43:01.979505644Z" level=info msg="API listen on /var/run/docker.sock"
```

The example contains entries from several different daemons and subsystems: networking, NFS, `cron`, Docker, and the power management daemon, `acpid`. Each message contains the following space-separated fields:

- Time stamp
- System’s hostname, in this case jessie
- Name of the process and its PID in square brackets
- Message payload

Some daemons encode the payload to add metadata about the message. In the output above, the `docker` process includes its own time stamp, a log level, and information about the configuration of the daemon itself. This additional information is entirely up to the sending process to generate and format.
**Rsyslog architecture**

Think about log messages as a stream of events and rsyslog as an event-stream processing engine. Log message “events” are submitted as inputs, processed by filters, and forwarded to output destinations. In rsyslog, each of these stages is configurable and modular. By default, rsyslog is configured in `/etc/rsyslog.conf`.

The **rsyslogd** process typically starts at boot and runs continuously. Programs that are syslog aware write log entries to the special file `/dev/log`, a UNIX domain socket. In a stock configuration for systems without **systemd**, **rsyslogd** reads messages from this socket directly, consults its configuration file for guidance on how to route them, and dispatches each message to an appropriate destination. It’s also possible (and common) to configure **rsyslogd** to listen for messages on a network socket.

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*See [this page](#) for more information about signals.*

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If you modify `/etc/rsyslog.conf` or any of its included files, you must restart the **rsyslogd** daemon to make your changes take effect. A TERM signal makes the daemon exit. A HUP signal causes **rsyslogd** to close all open log files, which is useful for rotating (renaming and restarting) logs.

By longstanding convention, **rsyslogd** writes its process ID to `/var/run/syslogd.pid`, so it’s easy to send signals to **rsyslogd** from a script. (On modern Linux systems, `/var/run` is a symbolic link to `/run`.) For example, the following command sends a hangup signal:

```
$ sudo kill -HUP '/bin/cat /var/run/syslogd.pid'
```

Trying to compress or rotate a log file that **rsyslogd** has open for writing is not healthy and has unpredictable results, so be sure to send a HUP signal before you do this. Refer to [this page](#) for information on sane log rotation with the **logrotate** utility.
**Rsyslog versions**

Red Hat and CentOS use rsyslog version 7, but Debian and Ubuntu have updated to version 8. FreeBSD users installing from ports can choose either version 7 or version 8. As you might expect, the rsyslog project recommends using the most recent version, and we defer to their advice. That said, it won’t make or break your logging experience if your operating system of choice is a version behind the latest and greatest.

Rsyslog 8 is a major rewrite of the core engine, and although a lot has changed under the hood for module developers, the user-facing aspects remain mostly unchanged. With a few exceptions, the configurations in the following sections are valid for both versions.
Rsyslog configuration

rsyslogd’s behavior is controlled by the settings in /etc/rsyslog.conf. All our example Linux distributions include a simple configuration with sensible defaults that suit most sites. Blank lines and lines beginning with a # are ignored. Lines in an rsyslog configuration are processed in order from beginning to end, and order is significant.

At the top of the configuration file are global properties that configure the daemon itself. These lines specify which input modules to load, the default format of messages, ownerships and permissions of files, the working directory in which to maintain rsyslog’s state, and other settings. The following example configuration is adapted from the default rsyslog.conf on Debian Jessie:

```
# Support local system logging
$ModLoad imuxsock

# Support kernel logging
$ModLoad imklog

# Write messages in the traditional time stamp format
$ActionFileDefaultTemplate RSYSLOG_TraditionalFileFormat

# New log files are owned by root:adm
$FileOwner root
$FileGroup adm

# Default permissions for new files and directories
$FileCreateMode 0640
$DirCreateMode 0755
$Umask 0022

# Location in which to store rsyslog working files
$WorkDirectory /var/spool/rsyslog
```

Most distributions use the $IncludeConfig legacy directive to include additional files from a configuration directory, typically /etc/rsyslog.d/*.conf. Because order is important, distributions organize files by preceding file names with numbers. For example, the default Ubuntu configuration includes the following files:

```
20-ufw.conf
21-cloudinit.conf
50-default.conf
```

rsyslogd interpolates these files into /etc/rsyslog.conf in lexicographic order to form its final configuration.

Filters, sometimes called “selectors,” constitute the bulk of an rsyslog configuration. They define how rsyslog sorts and processes messages. Filters are formed from expressions that select specific message criteria and actions that route selected messages to a desired destination.
Rsyslog understands three configuration syntaxes:

- Lines that use the format of the original syslog configuration file. This format is now known as “sysklogd format,” after the kernel logging daemon sysklogd. It’s simple and effective but has some limitations. Use it to construct simple filters.

- Legacy rsyslog directives, which always begin with a $ sign. The syntax comes from ancient versions of rsyslog and really ought to be obsolete. However, not all options have been converted to the newer syntax, and so this syntax remains authoritative for certain features.

- RainerScript, named for Rainer Gerhards, the lead author of rsyslog. This is a scripting syntax that supports expressions and functions. You can use it to configure most—but not all—aspects of rsyslog.

Many real-world configurations include a mix of all three formats, sometimes to confusing effect. Although it has been around since 2008, RainerScript remains slightly less common than the others. Fortunately, none of the dialects are particularly complex. In addition, many sites will have no need to do major surgery on the vanilla configurations included with their stock distributions.

To migrate from a traditional syslog configuration, simply start with your existing syslog.conf file and add options for the rsyslog features you want to activate.

Modules

Rsyslog modules extend the capabilities of the core processing engine. All inputs (sources) and outputs (destinations) are configured through modules, and modules can even parse and mutate messages. Although most modules were written by Rainer Gerhards, some were contributed by third parties. If you’re a C programmer, you can write your own.

Module names follow a predictable prefix pattern. Those beginning with im are input modules; om* are output modules, mm* are message modifiers, and so on. Most modules have additional configuration options that customize their behavior. The rsyslog module documentation is the complete reference.

The following list briefly describes some of the more common (or interesting) input and output modules, along with a few nuggets of exotica:

- imjournal integrates with the systemd journal, as described in Coexisting with syslog.

- imuxsock reads messages from a UNIX domain socket. This is the default when systemd is not present.

- imklog understands how to read kernel messages on Linux and BSD.
imfile converts a plain text file to syslog message format. It’s useful for importing log files generated by software that doesn’t have native syslog support. Two modes exist: polling mode, which checks the file for updates at a configurable interval, and notification mode (inotify), which uses the Linux filesystem event interface. This module is smart enough to resume where it left off whenever rsyslogd is restarted.

See this page for more information about TLS.

imtcp and imudp accept network messages over TCP and UDP, respectively. They allow you to centralize logging on a network. In combination with rsyslog’s network stream drivers, the TCP module can also accept mutually authenticated syslog messages through TLS. For Linux sites with extremely high volume, see also the imptcp module.

If the immark module is present, rsyslog produces time stamp messages at regular intervals. These time stamps can help you figure out that your machine crashed between 3:00 and 3:20 a.m., not just “sometime last night.” This information is also a big help when you are debugging problems that seem to occur regularly. Use the MarkMessagePeriod option to configure the mark interval.

omfile writes messages to a file. This is the most commonly used output module, and the only one configured in a default installation.

omfwd forwards messages to a remote syslog server over TCP or UDP. This is the module you’re looking for if your site needs centralized logging.

omkafka is a producer implementation for the Apache Kafka data streaming engine. Users at high-volume sites may benefit from being able to process messages that have many potential consumers.

Similarly to omkafka, omelasticsearch writes directly to an Elasticsearch cluster. See this page for more information about the ELK log management stack, which includes Elasticsearch as one of its components.

ommysql sends messages to a MySQL database. The rsyslog source distribution includes an example schema. Combine this module with the $MainMsgQueueSize legacy directive for better reliability.

Modules can be loaded and configured through either the legacy or RainerScript configuration formats. We show some examples in the format-specific sections below.

sysklogd syntax

The sysklogd syntax is the traditional syslog configuration format. If you encounter a standard syslogd, such as the version installed on stock FreeBSD, this is likely all you’ll need to
understand. (But note that the configuration file for the traditional **syslogd** is `/etc/syslog.conf`, not `/etc/rsyslog.conf`.)

This format is primarily intended for routing messages of a particular type to a desired destination file or network address. The basic format is

`selector   action`

The selector is separated from the action by one or more spaces or tabs. For example, the line

`auth.*   /var/log/auth.log`

causes messages related to authentication to be saved in `/var/log/auth.log`.

Selectors identify the source program (“facility”) that is sending a log message and the message’s priority level (“severity”) with the syntax

`facility.severity`

Both facility names and severity levels must be chosen from a short list of defined values; programs can’t make up their own. Facilities are defined for the kernel, for common groups of utilities, and for locally written programs. Everything else is classified under the generic facility “user.”

Selectors can contain the special keywords `*` and `none`, meaning all or nothing, respectively. A selector can include multiple facilities separated by commas. Multiple selectors can be combined with semicolons.

In general, selectors are ORed together: a message matching any selector is subject to the line’s `action`. However, a selector with a level of `none` excludes the listed facilities regardless of what other selectors on the same line might say.

Here are some examples of ways to format and combine selectors:

```plaintext
# Apply action to everything from facility.level
facility.level   action

# Everything from facility1.level and facility2.level
facility1, facility2.level   action

# Only facility1.level1 and facility2.level2
facility1.level1; facility2.level2 action

# All facilities with severity level
*.level   action

# All facilities except badfacility
*.level;badfacility.none   action
```

Table 10.2 lists the valid facility names. They are defined in **syslog.h** in the standard library.
Don’t take the distinction between auth and authpriv too seriously. All authorization-related messages are sensitive, and none should be world-readable. `sudo` logs use authpriv.

Table 10.3 lists the valid severity levels in order of descending importance.

The severity level of a message specifies its importance. The distinctions between the various levels are sometimes fuzzy. There’s a clear difference between notice and warning and between warning and err, but the exact shade of meaning expressed by alert as opposed to crit is a matter of conjecture.

Levels indicate the *minimum* importance that a message must have to be logged. For example, a message from SSH at level warning would match the selector `auth.warning` as well as the selectors...
auth.info, auth.notice, auth.debug, *.warning, *.notice, *.info, and *.debug. If the configuration directs auth.info messages to a particular file, auth.warning messages will go there also.

The format also allows the characters = and ! to be prefixed to priority levels to indicate “this priority only” and “except this priority and higher,” respectively. Table 10.4 shows examples.

Table 10.4: Examples of priority level qualifiers

<table>
<thead>
<tr>
<th>Selector</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>auth.info</td>
<td>Auth-related messages of info priority and higher</td>
</tr>
<tr>
<td>auth.=info</td>
<td>Only messages at info priority</td>
</tr>
<tr>
<td>auth.info;auth.lerr</td>
<td>Only priorities info, notice, and warning</td>
</tr>
<tr>
<td>auth.debug;auth.!=warning</td>
<td>All priorities except warning</td>
</tr>
</tbody>
</table>

The action field tells what to do with each message. Table 10.5 lists the options.

Table 10.5: Common actions

<table>
<thead>
<tr>
<th>Action</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>filename</td>
<td>Appends the message to a file on the local machine</td>
</tr>
<tr>
<td>@hostname</td>
<td>Forwards the message to the <code>rsyslogd</code> on hostname</td>
</tr>
<tr>
<td>@ipaddress</td>
<td>Forwards the message to <code>ipaddress</code> on UDP port 514</td>
</tr>
<tr>
<td>@@ipaddress</td>
<td>Forwards the message to <code>ipaddress</code> on TCP port 514</td>
</tr>
<tr>
<td></td>
<td>fifo name</td>
</tr>
<tr>
<td>user1,user2,.</td>
<td>Writes the message to the screens of <code>users</code> if they are logged in</td>
</tr>
<tr>
<td>*</td>
<td>Writes the message to all users who are currently logged in</td>
</tr>
<tr>
<td>~</td>
<td>Discards the message</td>
</tr>
<tr>
<td><code>^program;template</code></td>
<td>Formats the message according to the template specification and sends it to <code>program</code> as the first argument(^b)</td>
</tr>
</tbody>
</table>

\(^a\) See `man mkfifo` for more information.
\(^b\) See `man 5 rsyslog.conf` for further details on templates.

If a filename (or fifo name) action is specified, the name should be an absolute path. If you specify a nonexistent filename, `rsyslogd` will create the file when a message is first directed to it. The ownership and permissions of the file are specified in the global configuration directives as shown on this page.

Here are a few configuration examples that use the traditional syntax:
You can preface a `filename` action with a dash to indicate that the filesystem should not be synced after each log entry is written. syncing helps preserve as much logging information as possible in the event of a crash, but for busy log files it can be devastating in terms of I/O performance. We recommend including the dashes (and thereby inhibiting syncing) as a matter of course. Remove the dashes only temporarily when investigating a problem that is causing kernel panics.

**Legacy directives**

Although rsyslog calls these “legacy” options, they remain in widespread use, and you will find them in the majority of rsyslog configurations. Legacy directives can configure all aspects of rsyslog, including global daemon options, modules, filtering, and rules.

In practice, however, these directives are most commonly used to configure modules and the rsyslogd daemon itself. Even the rsyslog documentation warns against using the legacy format for message-processing rules, claiming that it is “extremely hard to get right.” Stick with the sysklogd or RainerScript formats for actually filtering and processing messages.

### Legacy directives

<table>
<thead>
<tr>
<th>Legacy directive</th>
<th>Destination</th>
</tr>
</thead>
<tbody>
<tr>
<td>kern.*</td>
<td>-/var/log/kern.log</td>
</tr>
<tr>
<td>cron.*</td>
<td>/var/log/cron.log</td>
</tr>
<tr>
<td>auth,authpriv.*</td>
<td>/var/log/auth.log</td>
</tr>
<tr>
<td><em>.::</em>;auth,authpriv,cron,kern.none</td>
<td>-/var/log/syslog</td>
</tr>
</tbody>
</table>

### Example Configuration

```plaintext
# Kernel messages to kern.log
kern.* -/var/log/kern.log
# Cron messages to cron.log
cron.* /var/log/cron.log
# Auth messages to auth.log
auth,authpriv.* /var/log/auth.log
# All other messages to syslog
*::*;auth,authpriv,cron,kern.none -/var/log/syslog
```

To put these options into effect, you could add the lines to a new file to be included in the main configuration such as `/etc/rsyslog.d/10-network-inputs.conf`. Then restart rsyslogd. Any options that modify a module’s behavior must appear after the module has been loaded.

**Table 10.6** describes a few of the more common legacy directives.
RainerScript

The RainerScript syntax is an event-stream-processing language with filtering and control-flow capabilities. In theory, you can also set basic `rsyslogd` options through RainerScript. But since some legacy options still don’t have RainerScript equivalents, why confuse things by using multiple option syntaxes?

RainerScript is more expressive and human-readable than `rsyslogd`’s legacy directives, but it has an unusual syntax that’s unlike any other configuration system we’ve seen. In practice, it feels somewhat cumbersome. Nonetheless, we recommend it for filtering and rule development if you need those features. In this section we discuss only a subset of its functionality.

Of our example distributions, only Ubuntu uses RainerScript in its default configuration files. However, you can use RainerScript format on any system running rsyslog version 7 or newer.

You can set global daemon parameters by using the `global()` configuration object. For example:

```plaintext
global(
    workDirectory="/var/spool/rsyslog"
    maxMessageSize="8192"
)
```

Most legacy directives have identically named RainerScript counterparts, such as `workDirectory` and `maxMessageSize` in the lines above. The equivalent legacy syntax for this configuration would be:

```plaintext
$WorkDirectory /var/spool/rsyslog
$MaxMessageSize 8192
```

You can also load modules and set their operating parameters through RainerScript. For example, to load the UDP and TCP modules and apply the same configuration demonstrated on this page, you’d use the following RainerScript:

<table>
<thead>
<tr>
<th>Option</th>
<th>Purpose</th>
</tr>
</thead>
<tbody>
<tr>
<td>$MainMsgQueueSize</td>
<td>Size of memory buffer between received and sent messages(^a)</td>
</tr>
<tr>
<td>$MaxMessageSize</td>
<td>Defaults to 8kB; must precede loading of any input modules</td>
</tr>
<tr>
<td>$LocalHostName</td>
<td>Overrides the local hostname</td>
</tr>
<tr>
<td>$WorkDirectory</td>
<td>Specifies where to save rsyslog working files</td>
</tr>
<tr>
<td>$ModLoad</td>
<td>Loads a module</td>
</tr>
<tr>
<td>$MaxOpenFiles</td>
<td>Modifies the defaults system nofile limit for <code>rsyslogd</code></td>
</tr>
<tr>
<td>$IncludeConfig</td>
<td>Includes additional configuration files</td>
</tr>
<tr>
<td>$UMASK</td>
<td>Sets the umask for new files created by <code>rsyslogd</code></td>
</tr>
</tbody>
</table>

---

\(^a\) This option is useful for slow outputs such as database inserts.
In RainerScript, modules have both “module parameters” and “input parameters.” A module is loaded only once, and a module parameter (e.g., the KeepAlive option in the imtcp module above) applies to the module globally. By contrast, input parameters can be applied to the same module multiple times. For example, we could instruct rsyslog to listen on both TCP ports 514 and 1514:

```
module(load="imtcp" KeepAlive="on")
input(type="imtcp" port="514")
input(type="imtcp" port="1514")
```

Most of the benefits of RainerScript relate to its filtering capabilities. You can use expressions to select messages that match a certain set of characteristics, then apply a particular action to the matching messages. For example, the following lines route authentication-related messages to /var/log/auth.log:

```
if $syslogfacility-text == 'auth' then {
   action(type="omfile" file="/var/log/auth.log")
}
```

In this example, $syslogfacility-text is a message property—that is, a part of the message’s metadata. Properties are prefixed by a dollar sign to indicate to rsyslog that they are variables. In this case, the action is to use the omfile output module to write matching messages to auth.log.

Table 10.7 lists some of the most frequently used properties.

<table>
<thead>
<tr>
<th>Property</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>$msg</td>
<td>The text of the message, without metadata</td>
</tr>
<tr>
<td>$rawmsg</td>
<td>The full message as received, including metadata</td>
</tr>
<tr>
<td>$hostname</td>
<td>The hostname from the message</td>
</tr>
<tr>
<td>$syslogfacility</td>
<td>Syslog facility in numerical form; see RFC3164</td>
</tr>
<tr>
<td>$syslogfacility-text</td>
<td>Syslog facility in text form</td>
</tr>
<tr>
<td>$syslogseverity</td>
<td>Syslog severity in numeric form; see RFC3164</td>
</tr>
<tr>
<td>$syslogseverity-text</td>
<td>Syslog severity in text form</td>
</tr>
<tr>
<td>$timegenerated</td>
<td>Time at which the message was received by rsyslogd</td>
</tr>
<tr>
<td>$timereported</td>
<td>Time stamp from the message itself</td>
</tr>
</tbody>
</table>

A given filter can include multiple filters and multiple actions. The following fragment targets kernel messages of critical severity. It logs the messages to a file and sends email to alert an
administrator of the problem.

module(load="ommmail")

if $syslogseverity-text == 'crit' and $syslogfacility-text == 'kern' then {
    action(type="omfile" file="/var/log/kern-crit.log")
    action(type="ommmail"
        server="smtp.admin.com"
        port="25"
        mailfrom="rsyslog@admin.com"
        mailto="ben@admin.com"
        subject.text="Critical kernel error"
        action.executeonlyonceeveryinterval="3600"
    )
}

Here, we’ve specified that we don’t want more than one email message generated per hour (3,600 seconds).

Filter expressions support regular expressions, functions, and other sophisticated techniques. Refer to the RainerScript documentation for complete details.
Config file examples

In this section we show three sample configurations for rsyslog. The first is a basic but complete configuration that writes log messages to files. The second example is a logging client that forwards syslog messages and httpd access and error logs to a central log server. The final example is the corresponding log server that accepts log messages from a variety of logging clients.

These examples rely heavily on RainerScript because it’s the suggested syntax for the latest versions of rsyslog. A few of the options are valid only in rsyslog version 8 and include Linux-specific settings such as inotify.

Basic rsyslog configuration

The following file can serve as a generic RainerScript rsyslog.conf for any Linux system:

```bash
module(load="imuxsock") # Local system logging
module(load="imklog") # Kernel logging
module(load="immark" interval="3600") # Hourly mark messages

# Set global rsyslogd parameters
global(
   workDirectory = "/var/spool/rsyslog"
   maxMessageSize = "8192"
)

# The output file module does not need to be explicitly loaded,
# but we can load it ourselves to override default parameter values.
module(load="builtin:omfile"
   # Use traditional timestamp format
   template="RSYSLOG TraditionalFileFormat"
)

   # Set the default permissions for all log files.
   fileOwner="root"
   fileGroup="adm"
   dirOwner="root"

$IncludeConfig /etc/rsyslog.d/*.conf

# Include files from /etc/rsyslog.d; there’s no RainerScript equivalent
)

   dirCreateMode="0755"
   fileCreateMode="0640"
   dirGroup="adm"
```

This example begins with a few default log collection options for rsyslogd. The default file permissions of 0640 for new log files is more restrictive than the omfile default of 0644.
**Network logging client**

This logging client forwards system logs and the Apache access and error logs to a remote server over TCP.

```conf
# Send all syslog messages to the server; this is sysklogd syntax
*.*   @logs.admin.com

# imfile reads messages from a file
# inotify is more efficient than polling
# It's the default, but noted here for illustration
module(load="imfile" mode="inotify")

# Import Apache logs through the imfile module
input(type="imfile"
    Tag="apache-access"
    File="/var/log/apache2/access.log"
    Severity="info"
)
input(type="imfile"
    Tag="apache-error"
    File="/var/log/apache2/error.log"
    Severity="info"
)

# Send Apache logs to the central log host
if $programname contains 'apache' then {
    action(type="omfwd"
        Target="logs.admin.com"
        Port="514"
        Protocol="tcp"
    )
}
```

Apache **httpd** does not write messages to syslog by default, so the access and error logs are read from text files with *imfile*. The messages are tagged for later use in a filter expression. (**httpd** can log directly to syslog with **mod_syslog**, but we use *imfile* here for illustration.)

At the end of the file, the `if` statement is a filter expression that searches for Apache messages and forwards those to logs.admin.com, the central log server. Logs are sent over TCP, which although more reliable than UDP still can potentially drop messages. You can use RELP (the Reliable Event Logging Protocol), a nonstandard output module, to guarantee log delivery.

See *Chapter 23* for more about configuration management.

In a real-world scenario, you might render the Apache-related portion of this configuration to `/etc/rsyslog.d/55-apache.conf` as part of the configuration management setup for the server.
Central logging host

The configuration of the corresponding central log server is straightforward: listen for incoming logs on TCP port 514, filter by log type, and write to files in the site-wide logging directory.

```
# Load the TCP input module and listen on port 514
# Do not accept more than 500 simultaneous clients
module(load="imtcp" MaxSessions="500")
input(type="imtcp" port="514")

# Save to different files based on the type of message
if $programname == 'apache-access' then {
    action(type="omfile" file="/var/log/site/apache/access.log")
} else if $programname == 'apache-error' then {
    action(type="omfile" file="/var/log/site/apache/error.log")
} else {
    # Everything else goes to a site-wide syslog file
    action(type="omfile" file="/var/log/site/syslog")
}
```

The central logging host generates a time stamp for each message as it writes out the message. Apache messages include a separate time stamp that was generated when `httpd` logged the message. You’ll find both of these time stamps in the site-wide log files.
Syslog message security

Rsyslog can send and receive log messages over TLS, a layer of encryption and authentication that runs on top of TCP. See this page for general information about TLS.

The example below assumes that the certificate authority, public certificates, and keys have already been generated. See this page for details on public key infrastructure and certificate generation.

This configuration introduces a new option: the network stream driver, a module that operates at a layer between the network and rsyslog. It typically implements features that enhance basic network capabilities. TLS is enabled by the gtls netstream driver.

The following example enables the gtls driver for a log server. The gtls driver requires a CA certificate, a public certificate, and the server’s private key. The imtcp module then enables the gtls stream driver.

```plaintext
global(
    defaultNetstreamDriver="gtls"
    defaultNetstreamDriverCAFile="/etc/ssl/ca/admin.com.pem"
    defaultNetstreamDriverCertFile="/etc/ssl/certs/server.com.pem"
    defaultNetstreamDriverKeyFile="/etc/ssl/private/server.com.key"
)
module(
    load="imtcp"
    streamDriver.name="gtls"
    streamDriver.mode="1"
    streamDriver.authMode="x509/name"
)
input(type="imtcp" port="6514")
```

The log server listens on the TLS version of the standard syslog port, 6514. The authMode option tells syslog what type of validation to perform. x509/name, the default, checks that the certificate is signed by a trusted authority and also validates the subject name that binds a certificate to a specific client through DNS.

Configuration for the client side of the TLS connection is similar. Use the client certificate and private key, and use the gtls netstream driver for the log forwarding output module.
In this case, we forward all log messages with a sort of Frankenstein version of the `sysklogd` syntax: the action component is a RainerScript form instead of one of the standard `sysklogd`-native options. If you need to be pickier about which messages to forward (or you need to send different classes of message to different destinations), you can use RainerScript filter expressions, as demonstrated in several of the examples earlier in this chapter.

```plaintext

global(
    defaultNetstreamDriver="gtls",
    defaultNetstreamDriverCAFile="/etc/ssl/ca/admin.com.pem",
    defaultNetstreamDriverCertFile="/etc/ssl/certs/client.com.pem",
    defaultNetstreamDriverKeyFile="/etc/ssl/private/client.com.key"
)

*.*    action(type="omfwd"
    Protocol="tcp"
    Target="logs.admin.com"
    Port="6514"
    StreamDriverMode="1"
    StreamDriver="gtls"
    StreamDriverAuthMode="x509/name"
)
```
Syslog configuration debugging

The **logger** command is useful for submitting log entries from shell scripts or the command line. You can also use it to test changes to rsyslog’s configuration. For example, if you have just added the line

```
local5.warning /tmp/evi.log
```

and want to verify that it is working, run the command

```
$ logger -p local5.warning "test message"
```

A line containing “test message” should be written to `/tmp/evi.log`. If this doesn’t happen, perhaps you’ve forgotten to restart **rsyslogd**?
10.4 Kernel and boot-time logging

The kernel and the system startup scripts present some special challenges in the domain of logging. In the case of the kernel, the problem is to create a permanent record of the boot process and kernel operation without building in dependencies on any particular filesystem or filesystem organization. For startup scripts, the challenge is to capture a coherent and accurate narrative of the startup process without permanently tying any system daemons to a startup log file, interfering with any programs’ own logging, or gooping up the startup scripts with glue that serves only to capture boot-time messages.

For kernel logging at boot time, kernel log entries are stored in an internal buffer of limited size. The buffer is large enough to accommodate messages about all the kernel’s boot-time activities. When the system is up and running, a user process accesses the kernel’s log buffer and disposes of its contents.

On Linux systems, `systemd-journald` reads kernel messages from the kernel buffer by reading the device file `/dev/kmsg`. You can view these messages by running `journalctl -k` or its alias, `journalctl --dmesg`. You can also use the traditional `dmesg` command.

On FreeBSD and older Linux systems, the `dmesg` command is the best way to view the kernel buffer; the output even contains messages that were generated before `init` started.

Another issue related to kernel logging is the appropriate management of the system console. As the system is booting, it’s important for all output to come to the console. However, once the system is up and running, console messages may be more an annoyance than a help, especially if the console is used for logins.

Under Linux, `dmesg` lets you set the kernel’s console logging level with a command-line flag. For example,

```
ubuntu$ sudo dmesg -n 2
```

Level 7 is the most verbose and includes debugging information. Level 1 includes only panic messages; the lower-numbered levels are the most severe. All kernel messages continue to go to the central buffer (and thence, to syslog) regardless of whether they are forwarded to the console.
10.5 Management and Rotation of Log Files

Erik Troan’s logrotate utility implements a variety of log management policies and is standard on all our example Linux distributions. It also runs on FreeBSD, but you’ll have to install it from the ports collection. By default, FreeBSD uses a different log rotation package, called newsyslog; see this page for details.
A logrotate configuration consists of a series of specifications for groups of log files to be managed. Options that appear outside the context of a log file specification (such as errors, rotate, and weekly in the following example) apply to all subsequent specifications. They can be overridden within the specification for a particular file and can also be respecified later in the file to modify the defaults.

Here’s a somewhat contrived example that handles several different log files:

```
# Global options
errors errors@book.admin.com
rotate 5
weekly

# Logfile rotation definitions and options
/var/log/messages {
    postrotate
        /bin/kill -HUP 'cat /var/run/syslogd.pid'
    endscript
}

/var/log/samba/*.log {
    notifempty
copytruncate
    sharedscripts
    postrotate
        /bin/kill -HUP 'cat /var/lock/samba/*.pid'
    endscript
}
```

This configuration rotates /var/log/messages every week. It keeps five versions of the file and notifies rsyslog each time the file is reset. Samba log files (there might be several) are also rotated weekly, but instead of being moved aside and restarted, they are copied and then truncated. The Samba daemons are sent HUP signals only after all log files have been rotated.

Table 10.8 lists the most useful logrotate.conf options.

**Table 10.8: logrotate options**
logrotate is normally run out of cron once a day. Its standard configuration file is /etc/logrotate.conf, but multiple configuration files (or directories containing configuration files) can appear on logrotate’s command line.

This feature is used by Linux distributions, which define the /etc/logrotate.d directory as a standard place for logrotate config files. logrotate-aware software packages (there are many) can drop in log management instructions as part of their installation procedure, thus greatly simplifying administration.

The delaycompress option is worthy of further explanation. Some applications continue to write to the previous log file for a bit after it has been rotated. Use delaycompress to defer compression for one additional rotation cycle. This option results in three types of log files lying around: the active log file, the previously rotated but not yet compressed file, and compressed, rotated files.

In addition to logrotate, Ubuntu has a simpler program called savelog that manages rotation for individual files. It’s more straightforward than logrotate and doesn’t use (or need) a config file. Some packages prefer to use their own savelog configurations rather than logrotate.
newsyslog: log management on FreeBSD

The misleadingly named `newsyslog`—so named because it was originally intended to rotate files managed by syslog—is the FreeBSD equivalent of `logrotate`. Its syntax and implementation are entirely different from those of `logrotate`, but aside from its peculiar date formatting, the syntax of a `newsyslog` configuration is actually somewhat simpler.

The primary configuration file is `/etc/newsyslog.conf`. See `man newsyslog` for the format and syntax. The default `/etc/newsyslog.conf` has examples for standard log files.

Like `logrotate`, `newsyslog` runs from `cron`. In a vanilla FreeBSD configuration, `/etc/crontab` includes a line that runs `newsyslog` once per hour.
10.6 MANAGEMENT OF LOGS AT SCALE

It’s one thing to capture log messages, store them on disk, and forward them to a central server. It’s another thing entirely to handle logging data from hundreds or thousands of servers. The message volumes are simply too high to be managed effectively without tools designed to function at this scale. Fortunately, multiple commercial and open source tools are available to address this need.
The ELK stack

The clear leader in the open source space—and indeed, one of the better software suites we’ve had the pleasure of working with—is the formidable “ELK” stack consisting of Elasticsearch, Logstash, and Kibana. This combination of tools helps you sort, search, analyze, and visualize large volumes of log data generated by a global network of logging clients. ELK is built by Elastic (elastic.co), which also offers training, support, and enterprise add-ons for ELK.

Elasticsearch is a scalable database and search engine with a RESTful API for querying data. It’s written in Java. Elasticsearch installations can range from a single node that handles a low volume of data to several dozen nodes in a cluster that indexes many thousands of events each second. Searching and analyzing log data is one of the most popular applications for Elasticsearch.

If Elasticsearch is the hero of the ELK stack, Logstash is its sidekick and trusted partner. Logstash accepts data from many sources, including queueing systems such as RabbitMQ and AWS SQS. It can also read data directly from TCP or UDP sockets and from the traditional logging stalwart, syslog. Logstash can parse messages to add additional structured fields and can filter out unwanted or nonconformant data. Once messages have been ingested, Logstash can write them to a wide variety of destinations, including, of course, Elasticsearch.

You can send log entries to Logstash in a variety of ways. You can configure a syslog input for Logstash and use the rsyslog omfwd output module, as described in Rsyslog configuration. You can also use a dedicated log shipper. Elastic’s own version is called Filebeat and can ship logs either to Logstash or directly to Elasticsearch.

The final ELK component, Kibana, is a graphical front end for Elasticsearch. It gives you a search interface through which to find the entries you need among all the data that has been indexed by Elasticsearch. Kibana can create graphs and visualizations that help to generate new insights about your applications. It’s possible, for example, to plot log events on a map to see geographically what’s happening with your systems. Other plug-ins add alerting and system monitoring interfaces.

Of course, ELK doesn’t come without operational burden. Building a large scale ELK stack with a custom configuration is no simple task, and managing it takes time and expertise. Most administrators we know (present company included!) have accidentally lost data because of bugs in the software or operational errors. If you choose to deploy ELK, be aware that you’re signing up for substantial administrative overhead.

We are aware of at least one service, logz.io, that offers production-grade ELK-as-a-service. You can send log messages from your network over encrypted channels to an endpoint that logz.io provides. There, the messages are ingested, indexed, and made available through Kibana. This is not a low-cost solution, but it’s worth evaluating. As with many cloud services, you may find that it’s ultimately more expensive to replicate the service locally.
Graylog

Graylog is the spunky underdog to ELK’s pack leader. It resembles the ELK stack in several ways: it keeps data in Elasticsearch, and it can accept log messages either directly or through Logstash, just as in the ELK stack. The real differentiator is the Graylog UI, which many users proclaim to be superior and easier to use.

See this page for more information about RBAC and this page for more information about LDAP.

Some of the enterprise (read: paid) features of ELK are included in the Graylog open source product, including support for role-based access control and LDAP integration. Graylog is certainly worthy of inclusion in a bake-off when you’re choosing a new logging infrastructure.
Logging as a service

Several commercial log management offerings are available. Splunk is the most mature and trusted; both hosted and on-premises versions are available. Some of the largest corporate networks rely on Splunk, not only as a log manager but also as a business analytics system. But if you choose Splunk, be prepared to pay dearly for the privilege.

Alternative SaaS options include Sumo Logic, Loggly, and Papertrail, all of which have native syslog integration and a reasonable search interface. If you use AWS, Amazon’s CloudWatch Logs service can collect log data both from AWS services and from your own applications.
10.7 Logging Policies

Over the years, log management has emerged from the realm of system administration minutiae to become a formidable enterprise management challenge in its own right. IT standards, legislative edicts, and provisions for security-incident handling may all impose requirements on the handling of log data. A majority of sites will eventually need to adopt a holistic and structured approach to the management of this data.

Log data from a single system has a relatively inconsequential effect on storage, but a centralized event register that covers hundreds of servers and dozens of applications is a different story entirely. Thanks in large part to the mission-critical nature of web services, application and daemon logs have become as important as those generated by the operating system.

Keep these questions in mind when designing your logging strategy:

- How many systems and applications will be included?
- What type of storage infrastructure is available?
- How long must logs be retained?
- What types of events are important?

The answers to these questions depend on business requirements and on any applicable standards or regulations. For example, one standard from the Payment Card Industry Security Standards Council requires that logs be retained on easy-access media (e.g., a locally mounted hard disk) for three months and archived to long-term storage for at least one year. The same standard also includes guidance about the types of data that must be included.

Of course, as one of our reviewers mentioned, you can’t be subpoenaed for log data you do not possess. Some sites do not collect (or intentionally destroy) sensitive log data for this reason. You might or might not be able get away with this kind of approach, depending on the compliance requirements that apply to you.

However you answer the questions above, be sure to gather input from your information security and compliance departments if your organization has them.

For most applications, consider capturing at least the following information:

- Username or user ID
- Event success or failure
- Source address for network events
- Date and time (from an authoritative source, such as NTP)
A log server should have a carefully considered storage strategy. For example, a cloud-based system might offer immediate access to 90 days of data, with a year of older data being rolled over to an object storage service and three additional years being saved to an archival storage solution. Storage requirements evolve over time, so a successful implementation must adapt easily to changing conditions.

Limit shell access to centralized log servers to trusted system administrators and personnel involved in addressing compliance and security issues. These log warehouse systems have no real role in the organization’s daily business beyond satisfying auditability requirements, so application administrators, end users, and the help desk have no business accessing them. Access to log files on the central servers should itself be logged.

Centralization takes work, and at smaller sites it may not represent a net benefit. We suggest twenty servers as a reasonable threshold for considering centralization. Below that size, just ensure that logs are rotated properly and are archived frequently enough to avoid filling up a disk. Include log files in a monitoring solution that alerts you if a log file stops growing.
11 Drivers and the Kernel

The kernel is the central government of a UNIX or Linux system. It’s responsible for enforcing
rules, sharing resources, and providing the core services that user processes rely on.

We don’t usually think too much about what the kernel is doing. That’s fortunate, because even a
simple command such as cat /etc/passwd entails a complex series of underlying actions. If the
system were an airliner, we’d want to think in terms of commands such as “increase altitude to
35,000 feet” rather than having to worry about the thousands of tiny internal steps that were
needed to manage the airplane’s control surfaces.

The kernel hides the details of the system’s hardware underneath an abstract, high-level interface.
It’s akin to an API for application programmers: a well-defined interface that provides useful
facilities for interacting with the system. This interface provides five basic features:

- Management and abstraction of hardware devices
- Processes and threads (and ways to communicate among them)
- Management of memory (virtual memory and memory-space protection)
- I/O facilities (filesystems, network interfaces, serial interfaces, etc.)
- Housekeeping functions (startup, shutdown, timers, multitasking, etc.)
Only device drivers are aware of the specific capabilities and communication protocols of the system’s hardware. User programs and the rest of the kernel are largely independent of that knowledge. For example, a filesystem on disk is very different from a network filesystem, but the kernel’s VFS layer makes them look the same to user processes and to other parts of the kernel. You don’t need to know whether the data you’re writing is headed to block 3,829 of disk device #8 or whether it’s headed for Ethernet interface e1000e wrapped in a TCP packet. All you need to know is that it will go to the file descriptor you specified.

Processes (and threads, their lightweight cousins) are the mechanisms through which the kernel implements CPU time sharing and memory protection. The kernel fluidly switches among the system’s processes, giving each runnable thread a small slice of time in which to get work done. The kernel prevents processes from reading and writing each other’s memory spaces unless they have explicit permission to do so.

The memory management system defines an address space for each process and creates the illusion that the process owns an essentially unlimited region of contiguous memory. In reality, different processes’ memory pages are jumbled together in the system’s physical memory. Only the kernel’s bookkeeping and memory protection schemes keep them sorted out.

Layered on top of the hardware device drivers, but below most other parts of the kernel, are the I/O facilities. These consist of filesystem services, the networking subsystem, and various other services that are used to get data into and out from the system.
Nearly all of the kernel’s multilayered functionality is written in C, with a few dabs of assembly language code thrown in to give access to CPU features that are not accessible through C compiler directives (e.g., the atomic read-modify-write instructions defined by many CPUs). Fortunately, you can be a perfectly effective system administrator without being a C programmer and without ever touching kernel code.

That said, it’s inevitable that at some point you’ll need to make some tweaks. These can take several forms.

Many of the kernel’s behaviors (such as network-packet forwarding) are controlled or influenced by tuning parameters that are accessible from user space. Setting these values appropriately for your environment and workload is a common administrative task.

Another common kernel-related task is the installation of new device drivers. New models and types of hardware (video cards, wireless devices, specialized audio cards, etc.) appear on the market constantly, and vendor-distributed kernels aren’t always equipped to take advantage of them.

In some cases, you may even need to build a new version of the kernel from source code. Sysadmins don’t have to build kernels as frequently as they used to, but it still makes sense in some situations. It’s easier than it sounds.

Kernels are tricky. It’s surprisingly easy to destabilize the kernel even through minor changes. Even if the kernel boots, it may not run as well as it should. What’s worse, you may not even realize that you’ve hurt performance unless you have a structured plan for assessing the results of your work. Be conservative with kernel changes, especially on production systems, and always have a backup plan for reverting to a known-good configuration.
11.2 Kernel version numbering

Before we dive into the depths of kernel wrangling, it’s worth spending a few words to discuss kernel versions and their relationship to distributions.

The Linux and FreeBSD kernels are under continuous active development. Over time, defects are fixed, new features added, and obsolete features removed.

Some older kernels continue to be supported for an extended period of time. Likewise, some distributions choose to emphasize stability and so run the older, more tested kernels. Other distributions try to offer the most recent device support and features but might be a bit less stable as a result. It’s incumbent upon you as an administrator to select among these options in a manner that accommodates your users’ needs. No single solution is appropriate for every context.
Linux kernel versions

The Linux kernel and the distributions based on it are developed separately from one other, so the kernel has its own versioning scheme. Some kernel releases do achieve a sort of iconic popularity, so it’s not unusual to find that several independent distributions are all using the same kernel. You can check with `uname -r` to see what kernel a given system is running.

See semver.org for more information about semantic versioning.

Linux kernels are named according to the rules of so-called semantic versioning, that is, they include three components: a major version, a minor version, and a patch level. At present, there is no predictable relationship between a version number and its intended status as a stable or development kernel; kernels are blessed as stable when the developers decide that they’re stable. In addition, the kernel’s major version number has historically been incremented somewhat capriciously.

Many stable versions of the Linux kernel can be under long-term maintenance at one time. The kernels shipped by major Linux distributions often lag the latest releases by a substantial margin. Some distributions even ship kernels that are formally out of date.

You can install newer kernels by compiling and installing them from the kernel source tree. However, we don’t recommend that you do this. Different distributions have different goals, and they select kernel versions appropriate to those goals. You never know when a distribution has avoided a newer kernel because of some subtle but specific concern. If you need a more recent kernel, install a distribution that’s designed around that kernel rather than trying to shoehorn the new kernel into an existing system.
FreeBSD kernel versions

FreeBSD takes a fairly straightforward approach to versions and releases. The project maintains two major production versions, which as of this writing are versions 10 and 11. The kernel has no separate versioning scheme; it’s released as part of the complete operating system and shares its version number.

The older of the two major releases (in this case, FreeBSD 10) can be thought of as a maintenance version. It doesn’t receive sweeping new features, and it’s maintained with a focus on stability and security updates.

The more recent version (FreeBSD 11, right now) is where active development occurs. Stable releases intended for general use are issued from this tree as well. However, the kernel code is always going to be newer and somewhat less battle-tested than that of the previous major version.

In general, dot releases occur about every four months. Major releases are explicitly supported for five years, and the dot releases within them are supported for three months after the next dot release comes out. That’s not an extensive lifetime for old dot releases; FreeBSD expects you to stay current with patches.
11.3 Devices and their drivers

A device driver is an abstraction layer that manages the system’s interaction with a particular type of hardware so that the rest of the kernel doesn’t need to know its specifics. The driver translates between the hardware commands understood by the device and a stylized programming interface defined (and used) by the kernel. The driver layer helps keep the majority of the kernel device-independent.

Given the remarkable pace at which new hardware is developed, it is practically impossible to keep main-line OS distributions up to date with the latest hardware. Hence, you will occasionally need to add a device driver to your system to support a new piece of hardware.

Device drivers are system-specific, and they are often specific to a particular range of kernel revisions as well. Drivers for other operating systems (e.g., Windows) do not work on UNIX and Linux, so keep this in mind when you purchase new hardware. In addition, devices vary in their degree of compatibility and functionality when used with various Linux distributions, so it’s wise to pay some attention to the experiences that other sites have had with any hardware you are considering.

Hardware vendors are attracted to the FreeBSD and Linux markets and often publish appropriate drivers for their products. In the optimal case, your vendor furnishes you with both a driver and installation instructions. Occasionally, you might find the driver you need only on some sketchy-looking and uncommented web page. Caveat emptor.
Device files and device numbers

In most cases, device drivers are part of the kernel; they are not user processes. However, a driver can be accessed both from within the kernel and from user space, usually through “device files” that live in the /dev directory. The kernel maps operations on these files into calls to the code of the driver.

Most non-network devices have one or more corresponding files in /dev. Complex servers may support hundreds of devices. By virtue of being device files, the files in /dev each have a major and minor device number associated with them. The kernel uses these numbers to map device-file references to the corresponding driver.

The major device number identifies the driver with which the file is associated (in other words, the type of device). The minor device number usually identifies which particular instance of a given device type is to be addressed. The minor device number is sometimes called the unit number.

You can see the major and minor number of a device file with ls -l:

```
linux$ ls -l /dev/sda
brw-rw---- 1 root disk 8, 0 Jul 13 01:38 /dev/sda
```

This example shows the first SCSI/SATA/SAS disk on a Linux system. It has a major number of 8 and a minor number of 0.

The minor device number is sometimes used by the driver to select or enable certain characteristics particular to that device. For example, a tape drive can have one file in /dev that rewinds the drive automatically when it’s closed and another file that does not. The driver is free to interpret the minor device number in whatever way it likes. Look up the man page for the driver to determine what convention it is using.

There are actually two types of device files: block device files and character device files. A block device is read or written one block (a group of bytes, usually a multiple of 512) at a time; a character device can be read or written one byte at a time. The character b at the start of the ls output above indicates that /dev/sda is a block device; ls would show this character as a c if it were a character device.

Traditionally, certain devices could act as either block or character devices, and separate device files existed to make them accessible in either mode. Disks and tapes led dual lives, but most other devices did not. However, this parallel access system is not used anymore. FreeBSD represents all formerly dual-mode devices as character devices, and Linux represents them as block devices.

It is sometimes convenient to implement an abstraction as a device driver even when it controls no actual device. Such phantom devices are known as pseudo-devices. For example, a user who
logs in over the network is assigned a pseudo-TTY (PTY) that looks, feels, and smells like a
serial port from the perspective of high-level software. This trick allows programs written in the
days when everyone used a physical terminal to continue to function in the world of windows
and networks. `/dev/zero`, `/dev/null`, and `/dev/urandom` are some other examples of pseudo-
devices.

When a program performs an operation on a device file, the kernel intercepts the reference, looks
up the appropriate function name in a table, and transfers control to the appropriate part of the
driver.

To perform an operation that doesn’t have a direct analog in the filesystem model (ejecting a
DVD, for example), a program traditionally uses the `ioctl` system call to pass a message directly
from user space into the driver. Standard `ioctl` request types are registered by a central authority
in a manner similar to the way that network protocol numbers are maintained by IANA.

FreeBSD continues to use the traditional `ioctl` system. Traditional Linux devices also use `ioctl`,
but modern networking code uses the more flexible Netlink sockets system described in
RFC3549. These sockets provide a more flexible messaging system than `ioctl` without the need
for a central authority.
Challenges of device file management

Device files have been a tricky problem for many years. When systems supported only a few types of devices, manual maintenance of device files was manageable. As the number of available devices grew, however, the /dev filesystem became cluttered, often with files irrelevant to the current system. Red Hat Enterprise Linux version 3 included more than 18,000 device files, one for every possible device that could be attached to the system! The creation of static device files quickly became a crushing problem and an evolutionary dead end.

USB, FireWire, Thunderbolt, and other device interfaces introduce additional wrinkles. Ideally, a drive that is initially recognized as /dev/sda would remain available as /dev/sda despite intermittent disconnections and regardless of the activity of other devices and buses. The presence of other transient devices such as cameras, printers, and scanners (not to mention other types of removable media) muddies the waters and makes the persistent identity problem even worse.

Network interfaces have this same problem; they are devices but do not have device files to represent them in /dev. For these devices, the modern approach is to use the relatively simple Predictable Network Interface Names system, which assigns interface names that are stable across reboots, changes in hardware, and changes in drivers. Modern systems now have analogous methods for dealing with the names of other devices, too.
Manual creation of device files

Modern systems manage their device files automatically. However, a few rare corner cases may still require you to create devices manually with the `mknod` command. So here’s how to do it:

```
 mknod filename type major minor
```

Here, `filename` is the device file to be created, `type` is `c` for a character device or `b` for a block device, and `major` and `minor` are the major and minor device numbers. If you are creating a device file that refers to a driver that’s already present in your kernel, check the documentation for the driver to find the appropriate major and minor device numbers.
Modern device file management

Linux and FreeBSD both automate the management of device files. In classic UNIX fashion, the systems are more or less the same in concept but entirely separate in their implementations and in the formats of their configuration files. Let a thousand flowers bloom!

When a new device is detected, both systems automatically create the device’s corresponding device files. When a device goes away (e.g., a USB thumb drive is unplugged), its device files are removed. For architectural reasons, both Linux and FreeBSD isolate the “creating device files” part of this equation.

In FreeBSD, devices are created by the kernel in a dedicated filesystem type (devfs) that’s mounted on `/dev`. In Linux, a daemon running in user space called `udev` is responsible for this activity. Both systems listen to an underlying stream of kernel-generated events that report the arrival and departure of devices.

However, there’s a lot more we might want to do with a newly discovered device than just create a device file for it. If it represents a piece of removable storage media, for example, we might want to automount it as a filesystem. If it’s a hub or a communications device, we might want to get it set up with the appropriate kernel subsystem.

Both Linux and FreeBSD leave such advanced procedures to a user-space daemon: `udevd` in the case of Linux, and `devd` in the case of FreeBSD. The main conceptual distinction between the two platforms is that Linux concentrates most intelligence in `udevd`, whereas FreeBSD’s devfs filesystem is itself slightly configurable.

Table 11.1 summarizes the components of the device file management systems on both platforms.

### Table 11.1: Outline of automatic device management

<table>
<thead>
<tr>
<th>Component</th>
<th>Linux</th>
<th>FreeBSD</th>
</tr>
</thead>
<tbody>
<tr>
<td><code>/dev</code> filesystem</td>
<td>udev / devtmpfs</td>
<td>devfs</td>
</tr>
<tr>
<td><code>/dev</code> FS configuration files</td>
<td>–</td>
<td><code>/etc/devfs.conf</code></td>
</tr>
<tr>
<td></td>
<td></td>
<td><code>/etc/devfs.rules</code></td>
</tr>
<tr>
<td>Device manager daemon</td>
<td>udevd</td>
<td>devd</td>
</tr>
<tr>
<td>Daemon configuration files</td>
<td><code>/etc/udev/udev.conf</code></td>
<td><code>/etc/devd.conf</code></td>
</tr>
<tr>
<td></td>
<td><code>/etc/udev/rules.d</code></td>
<td></td>
</tr>
<tr>
<td></td>
<td><code>/lib/udev/rules.d</code></td>
<td></td>
</tr>
<tr>
<td>Filesystem automounts</td>
<td>udevd</td>
<td>autosfs</td>
</tr>
</tbody>
</table>
Linux device management

Linux administrators should understand how udevd’s rule system works and should know how to use the udevadm command. Before peering into those details, however, let’s first review the underlying technology of sysfs, the device information repository from which udevd gets its raw data.

Sysfs: a window into the souls of devices

Sysfs was added to the Linux kernel at version 2.6. It is a virtual, in-memory filesystem implemented by the kernel to provide detailed and well-organized information about the system’s available devices, their configurations, and their state. Sysfs device information is accessible both from within the kernel and from user space.

You can explore the /sys directory, where sysfs is typically mounted, to find out everything from what IRQ a device is using to how many blocks have been queued for writing to a disk controller. One of the guiding principles of sysfs is that each file in /sys should represent only one attribute of the underlying device. This convention imposes a certain amount of structure on an otherwise chaotic data set.

Table 11.2 shows the directories within /sys, each of which is a subsystem that has been registered with sysfs. The exact directories vary slightly by distribution.

Table 11.2: Subdirectories of /sys

<table>
<thead>
<tr>
<th>Directory</th>
<th>What it contains</th>
</tr>
</thead>
<tbody>
<tr>
<td>block</td>
<td>Information about block devices such as hard disks</td>
</tr>
<tr>
<td>bus</td>
<td>Buses known to the kernel: PCI-E, SCSI, USB, and others</td>
</tr>
<tr>
<td>class</td>
<td>A tree organized by functional types of devices a</td>
</tr>
<tr>
<td>dev</td>
<td>Device information split between character and block devices</td>
</tr>
<tr>
<td>devices</td>
<td>An ancestrally correct representation of all discovered devices</td>
</tr>
<tr>
<td>firmware</td>
<td>Interfaces to platform-specific subsystems such as ACPI</td>
</tr>
<tr>
<td>fs</td>
<td>A directory for some, but not all, filesystems known to the kernel</td>
</tr>
<tr>
<td>kernel</td>
<td>Kernel internals such as cache and virtual memory status</td>
</tr>
<tr>
<td>module</td>
<td>Dynamic modules loaded by the kernel</td>
</tr>
<tr>
<td>power</td>
<td>A few details about the system’s power state; mostly unused</td>
</tr>
</tbody>
</table>

a. For example, sound and graphic cards, input devices, and network interfaces

Device configuration information was formerly found in the /proc filesystem, if it was available at all. /proc was inherited from System V UNIX and grew organically and somewhat randomly over time. It ended up collecting all manner of unrelated information, including many elements unrelated to processes. Although extra junk in /proc is still supported for backward
compatibility, /sys is a more predictable and organized way of reflecting the kernel’s internal data structures. We anticipate that all device-specific information will move to /sys over time.

**udevadm: explore devices**

The **udevadm** command queries device information, triggers events, controls the **udevd** daemon, and monitors udev and kernel events. Its primary use for administrators is to build and test rules, which are covered in the next section.

**udevadm** expects one of six commands as its first argument: **info**, **trigger**, **settle**, **control**, **monitor**, or **test**. Of particular interest to system administrators are **info**, which prints device-specific information, and **control**, which starts and stops **udevd** or forces it to reload its rules files. The **monitor** command displays events as they occur.

The following command shows all udev attributes for the device sdb. The output is truncated here, but in reality it goes on to list all parent devices—such as the USB bus—that are ancestors of sdb in the device tree.

```
linux$ udevadm info -a -n sdb
...
looking at device '/devices/pci0000:00/0000:00:11.0/0000:02:03.0/
  usb1/1-1/1-1:1.0/host6/target6:0:0/6:0:0:0/block/sdb':
  KERNEL="sdb"
  SUBSYSTEM="block"
  DRIVER=""
  ATTR{range}"16"
  ATTR{ext_range}"256"
  ATTR{removable}"1"
  ATTR{ro}"0"
  ATTR{size}"1974271"
  ATTR{capability}"53"
  ATTR{stat}" 71 986 1561 860 1 0 1 12 0 592 872"
...
```

All paths in **udevadm** output (such as `/devices/pci0000:00/0000:00:11.0/0000:02:03.0/`) are relative to /sys, even though they may appear to be absolute pathnames.

The output is formatted so that you can feed it back to udev when constructing rules. For example, if the `ATTR{size}="1974271"` clause were unique to this device, you could copy that snippet into a rule as the identifying criterion.

Refer to the man page on **udevadm** for additional options and syntax.

**Rules and persistent names**

**udevd** relies on a set of rules to guide its management of devices. The default rules live in the `/lib/udev/rules.d` directory, but local rules belong in `/etc/udev/rules.d`. You need never edit or
delete the default rules; you can ignore or override a file of default rules by creating a new file with the same name in the custom rules directory.

The master configuration file for udevd is /etc/udev/udev.conf; however, the default behaviors are reasonable. The udev.conf files on our example distributions contain only comments, with the exception of one line that enables error logging.

Sadly, because of political bickering among distributors and developers, there is little rule synergy among distributions. Many of the filenames in the default rules directory are the same from distribution to distribution, but the contents of the files differ significantly.

Rule files are named according to the pattern nn-description.rules, where nn is usually a two-digit number. Files are processed in lexical order, so lower numbers are processed first. Files from the two rules directories are combined before the udev daemon, udevd, parses them. The .rules suffix is mandatory; files without it are ignored.

Rules are of the form

\[\text{match\_clause}, \left[ \text{match\_clause}, \ldots \right] \text{assign\_clause}, \left[ \text{assign\_clause}, \ldots \right]\]

The match clauses define the situations in which the rule is to be applied, and the assignment clauses tell udevd what to do when a device is consistent with all the rule’s match clauses. Each clause consists of a key, an operator, and a value. For example, the match clause \text{ATTR\{size\}=="1974271"} was referred to above as a potential component of a rule; it selects all devices whose size attribute is exactly 1,974,271.

Most match keys refer to device properties (which udevd obtains from the /sys filesystem), but some refer to other context-dependent attributes, such as the operation being handled (e.g., device addition or removal). All match clauses must match in order for a rule to be activated.

Table 11.3 shows the match keys understood by udevd.

Table 11.3: udevd match keys
The assignment clauses specify actions udevd should take to handle any matching events. Their format is similar to that for match clauses.

The most important assignment key is NAME, which indicates how udevd should name a new device. The optional SYMLINK assignment key creates a symbolic link to the device through its desired path in /dev.

Here, we put these components together with an example configuration for a USB flash drive. Suppose we want to make the drive’s device name persist across insertions and we want the drive to be mounted and unmounted automatically.

To start with, we insert the flash drive and check to see how the kernel identifies it. This task can be approached in a couple of ways. By running the lsusb command, we can inspect the USB bus directly:

```
ubuntu$ lsusb
Bus 001 Device 007: ID 1307:0163 Transcend, Inc. USB Flash Drive
Bus 001 Device 001: ID 1d6b:0002 Linux Foundation 2.0 root hub
Bus 002 Device 001: ID 1d6b:0001 Linux Foundation 1.1 root hub
```

Alternatively, we can check for kernel log entries by running dmesg or journalctl. In our case, the attachment leaves an extensive audit trail:

---

<table>
<thead>
<tr>
<th>Match key</th>
<th>Function</th>
</tr>
</thead>
<tbody>
<tr>
<td>ACTION</td>
<td>Matches the event type, e.g., add or remove</td>
</tr>
<tr>
<td>ATTR{filename}</td>
<td>Matches a device’s sysfs values&lt;sup&gt;a&lt;/sup&gt;</td>
</tr>
<tr>
<td>DEVPATH</td>
<td>Matches a specific device path</td>
</tr>
<tr>
<td>DRIVER</td>
<td>Matches the driver used by a device</td>
</tr>
<tr>
<td>ENV{key}</td>
<td>Matches the value of an environment variable</td>
</tr>
<tr>
<td>KERNEL</td>
<td>Matches the kernel’s name for the device</td>
</tr>
<tr>
<td>PROGRAM</td>
<td>Runs an external command; matches if the return code is 0</td>
</tr>
<tr>
<td>RESULT</td>
<td>Matches the output of the last call through PROGRAM</td>
</tr>
<tr>
<td>SUBSYSTEM</td>
<td>Matches a specific subsystem</td>
</tr>
<tr>
<td>TEST{omask}</td>
<td>Tests whether a file exists; the omask is optional</td>
</tr>
</tbody>
</table>

---

<sup>a</sup> The filename is a leaf in the sysfs tree that corresponds to a specific attribute.
The log messages above indicate that the drive was recognized as sdb, which gives us an easy way to identify the device in /sys. We can now examine the /sys filesystem with udevadm in search of some rule snippets that are characteristic of the device and so might be useful to incorporate in udev rules.

```
ubuntu$ udevadm info -a -p /block/sdb/sdb1
looking at device '/devices/pcl0000:00/0000:00:11.0/0000:02:03.0/usb1/1-1/1-1:1.0/host30/target30:0:0:30:0:0:0:0:block/sdb1':
   KERNEL="sdb1"
   SUBSYSTEM="block"
   DRIVER=""
   ATTR[partition]="1"
   ATTR[start]="63"
   ATTR[size]="1974208"
   ATTR[stat]=" 71 792 1857 808 0 0 0 0 0 512 808"
looking at parent device '/devices/pcl0000:00/0000:00:11.0/0000:02:03.0.0/usb1/1-1/1-1:1.0/host30/target30:0:0:30:0:0:0:0/block/sdb':
   KERNEL="sdb"
   SUBSYSTEMS="block"
   DRIVERS=""
   ATTRS[scsi_level]="3"
   ATTRS[vendor]="Ut163"
   ATTRS[model]="USB2FlashStorage"
```

The output from udevadm shows several opportunities for matching. One possibility is the `size` field, which is likely to be unique to this device. However, if the size of the partition were to change, the device would not be recognized. Instead, we can use a combination of two values: the kernel’s naming convention of sd plus an additional letter, and the contents of the `model` attribute, `USB2FlashStorage`. For creating rules specific to this particular flash drive, another good choice would be the device’s serial number (which we’ve omitted from the output here).

We next put our rules for this device in the file `/etc/udev/rules.d/10-local.rules`. Because we have multiple objectives in mind, we need a series of rules.
First, we take care of creating device symlinks in /dev. The following rule uses our knowledge of the ATTRS and KERNEL match keys, gleaned from udevadm, to identify the device:

```plaintext
ATTRS{model}="USB2FlashStorage", KERNEL="sd[a-z]1",
SYMLINK+="ate-flash\%n"
```

(The rule has been folded here to fit the page; in the original file, it’s all one line.)

When the rule triggers, udevd sets up /dev/ate-flash\(N\) as a symbolic link to the device (where \(N\) is the next integer in sequence, starting at 0). We don’t really expect more than one of these devices to appear on the system. If more copies do appear, they receive unique names in /dev, but the exact names will depend on the insertion order of the devices.

Next, we use the ACTION key to run some commands whenever the device appears on the USB bus. The RUN assignment key lets us create an appropriate mount point directory and mount the device there.

```plaintext
ACTION="add", ATTRS{model}="USB2FlashStorage", KERNEL="sd[a-z]1",
    RUN="/bin/mkdir -p /mnt/ate-flash\%n"
ACTION="add", ATTRS{model}="USB2FlashStorage", KERNEL="sd[a-z]1",
    PROGRAM="/lib/udev/vol_id -t %N", RESULT="vfat",
    RUN="/bin/mount vfat /dev/%k /mnt/ate-flash\%n"
```

The PROGRAM and RUN keys look similar, but PROGRAM is a match key that’s active during the rule selection phase, whereas RUN is an assignment key that’s part of the rule’s actions once triggered. The second rule above verifies that the flash drive contains a Windows filesystem before mounting it with the -t vfat option to the mount command.

Similar rules clean up when the device is removed:

```plaintext
ACTION="remove", ATTRS{model}="USB2FlashStorage",
    KERNEL="sd[a-z]1", RUN="/bin/umount -l /mnt/ate-flash\%n"
ACTION="remove", ATTRS{model}="USB2FlashStorage",
    KERNEL="sd[a-z]1", RUN="/bin/rmdir /mnt/ate-flash\%n"
```

Now that our rules are in place, we must notify udevd of our changes. udevadm’s control command is one of the few that require root privileges:

```
ubuntu$ sudo udevadm control --reload-rules
```

Typos are silently ignored after a reload, even with the --debug flag, so be sure to double-check the rules’ syntax.

That’s it! Now when the flash drive is plugged into a USB port, udevd creates a symbolic link called /dev/ate-flash1 and mounts the drive as /mnt/ate-flash1.
ubuntu$ ls -l /dev/ate*
1rwxrwxrwx 1 root root 4 2009-08-09 21:22 /dev/ate-flash1 -> sdb1

ubuntu$ mount | grep ate
/dev/sdb1 on /mnt/ate-flash1 type vfat (rw)
FreeBSD device management

As we saw in the brief overview on this page, FreeBSD’s implementation of the self-managing /dev filesystem is called devfs, and its user-level device management daemon is called devd.

Devfs: automatic device file configuration

Unlike Linux’s udev filesystem, devfs itself is somewhat configurable. However, the configuration system is both peculiar and rather impotent. It’s split into boot-time (/etc/devfs.conf) and dynamic (/etc/devfs.rules) portions. The two configuration files have different syntaxes and somewhat different capabilities.

Devfs for static (nonremovable) devices is configured in /etc/devfs.conf. Each line is a rule that starts with an action. The possible actions are link, own, and perm. The link action sets up symbolic links for specific devices. The own and perm actions change the ownerships and permissions of device files, respectively.

Each action accepts two parameters, the interpretation of which depends on the specific action. For example, suppose we want our DVD-ROM drive /dev/cd0 to also be accessible by the name /dev/dvd. The following line would do the trick:

link cd0 dvd

We could set the ownerships and permissions on the device with the following lines.

own cd0 root:sysadmin
perm cd0 0660

Just as /etc/devfs.conf specifies actions to take for built-in devices, /etc/devfs.rules contains rules for removable devices. Rules in devfs.rules also have the option to make devices hidden or inaccessible, which can be useful for jail(8) environments.

devd: higher-level device management

The devd daemon runs in the background, watching for kernel events related to devices and acting on the rules defined in /etc/devd.conf. The configuration of devd is detailed in the devd.conf man page, but the default devd.conf file includes many useful examples and enlightening comments.

The format of /etc/devd.conf is conceptually simple, consisting of “statements” containing groups of “substatements”. Statements are essentially rules, and substatements provide details about the rule. Table 11.4 lists the available statement types.

Table 11.4: Statement types in /etc/devd.conf
Despite being conceptually simple, the configuration language for substatements is rich and complex. For this reason, many of the common configuration statements are already included in the standard distribution’s configuration file. In many cases, you will never need to modify the default `/etc/devd.conf`.

Automatic mounting of removable media devices such as USB hard disks and thumb drives is now handled by FreeBSD’s implementation of autofs, not by `devd`. See this page for general information about autofs. Although autofs is found on most UNIX-like operating systems, FreeBSD is unusual in assigning it this extra task.

<table>
<thead>
<tr>
<th>Statement</th>
<th>What it specifies</th>
</tr>
</thead>
<tbody>
<tr>
<td>attach</td>
<td>What to do when a device is inserted</td>
</tr>
<tr>
<td>detach</td>
<td>What to do when a device is removed</td>
</tr>
<tr>
<td>nomatch</td>
<td>What to do if no other statements match a device</td>
</tr>
<tr>
<td>notify</td>
<td>How to respond to kernel events about a device</td>
</tr>
<tr>
<td>options</td>
<td>Configuration options for <code>devd</code> itself</td>
</tr>
</tbody>
</table>

11.4 Linux kernel configuration

You can use any of three basic methods to configure a Linux kernel. Chances are that you will have the opportunity to try all of them eventually. The methods are

- Modifying tunable (dynamic) kernel configuration parameters
- Building a kernel from scratch (by compiling it from the source code, possibly with modifications and additions)
- Loading new drivers and modules into an existing kernel on the fly

These procedures are used in different situations, so learning which approaches are needed for which tasks is half the battle. Modifying tunable parameters is the easiest and most common kernel tweak, whereas building a kernel from source code is the hardest and least often required. Fortunately, all these approaches become second nature with a little practice.
Tuning Linux kernel parameters

Many modules and drivers in the kernel were designed with the knowledge that one size doesn’t fit all. To increase flexibility, special hooks allow parameters such as an internal table’s size or the kernel’s behavior in a particular circumstance to be adjusted on the fly by the system administrator. These hooks are accessible through an extensive kernel-to-userland interface represented by files in the /proc filesystem (aka procs). In some cases, a large user-level application (especially an infrastructure application such as a database) might require a sysadmin to adjust kernel parameters to accommodate its needs.

You can view and set kernel options at run time through special files in /proc/sys. These files mimic standard Linux files, but they are really back doors into the kernel. If a file in /proc/sys contains a value you want to change, you can try writing to it. Unfortunately, not all files are writable (regardless of their apparent permissions), and not much documentation is available. If you have the kernel source tree installed, you may be able to read about some of the values and their meanings in the subdirectory Documentation/sysctl (or on-line at kernel.org/doc).

For example, to change the maximum number of files the system can have open at once, try something like

```
linux# cat /proc/sys/fs/file-max
34916
linux# echo 32768 > /proc/sys/fs/file-max
```

Once you get used to this unorthodox interface, you’ll find it quite useful. However, note that changes are not remembered across reboots.

A more permanent way to modify these same parameters is to use the `sysctl` command. `sysctl` can set individual variables either from the command line or from a list of `variable=value` pairs in a configuration file. By default, /etc/sysctl.conf is read at boot time and its contents are used to set the initial values of parameters.

For example, the command

```
linux# sysctl net.ipv4.ip_forward=0
```

turns off IP forwarding. (Alternatively, you can manually edit /etc/sysctl.conf.) You form the variable names used by `sysctl` by replacing the slashes in the /proc/sys directory structure with dots.

Table 11.5 lists some commonly tuned parameters for Linux kernel version 3.10.0 and higher. Default values vary widely among distributions.

**Table 11.5: Files in /proc/sys for some tunable kernel parameters**
<table>
<thead>
<tr>
<th>File</th>
<th>What it does</th>
</tr>
</thead>
<tbody>
<tr>
<td>cdrom/autoclose</td>
<td>Autocloses the CD-ROM when mounted</td>
</tr>
<tr>
<td>cdrom/autoeject</td>
<td>Autoejects the CD-ROM when unmounted</td>
</tr>
<tr>
<td>fs/file-max</td>
<td>Sets max number of open files</td>
</tr>
<tr>
<td>kernel/ctrl-alt-del</td>
<td>Reboots on &lt;Control-Alt-Delete&gt;; may increase security on unsecured consoles</td>
</tr>
<tr>
<td>kernel/panic</td>
<td>Sets seconds to wait before rebooting after a kernel panic: 0 = loop or hang indefinitely</td>
</tr>
<tr>
<td>kernel/panic_on_oops</td>
<td>Determines the kernel’s behavior after encountering an oops or a bug: 1 = always panic</td>
</tr>
<tr>
<td>kernel/printk_ratelim</td>
<td>Sets minimum seconds between kernel messages</td>
</tr>
<tr>
<td>kernel/printk_ratelim_burst</td>
<td>Sets number of messages in succession before the printk rate limit is actually enforced</td>
</tr>
<tr>
<td>kernel/shmmax</td>
<td>Sets max amount of shared memory</td>
</tr>
<tr>
<td>net/ip*/conf/default/rp_filter</td>
<td>Enables IP source route verification</td>
</tr>
<tr>
<td>net/ip*/icmp_echo_ignore_all</td>
<td>Ignores ICMP pings when set to 1</td>
</tr>
<tr>
<td>net/ip*/ip_forward</td>
<td>Allows IP forwarding when set to 1</td>
</tr>
<tr>
<td>net/ip*/ip_local_port_range</td>
<td>Sets local port range used during connection setup</td>
</tr>
<tr>
<td>net/ip*/tcp_syncookies</td>
<td>Protects against SYN flood attacks; turn on if you suspect denial-of-service (DoS) attacks</td>
</tr>
<tr>
<td>tcp_fin_timeout</td>
<td>Sets seconds to wait for a final TCP FIN packet</td>
</tr>
<tr>
<td>vm/overcommit_memory</td>
<td>Controls memory overcommit behavior, i.e., how the kernel reacts when physical memory is insufficient to handle a VM allocation request</td>
</tr>
<tr>
<td>vm/overcommit_ratio</td>
<td>Defines how much physical memory (as a percentage) will be used when overcommitting</td>
</tr>
</tbody>
</table>

Note that there are two IP networking subdirectories of `/proc/sys/net`: **ipv4** and **ipv6**. In the past, administrators only had to worry about IPv4 behaviors because that was the only game in town. But as of this writing (2017), the IPv4 address blocks have all been assigned, and IPv6 is deployed and in use almost everywhere, even within smaller organizations.

In general, when you change a parameter for IPv4, you should also change that parameter for IPv6, if you are supporting both protocols. It’s all too easy to modify one version of IP and not the other, then run into problems several months or years later when a user reports strange network behavior.
Building a custom kernel

Because Linux evolves rapidly, you’ll likely be faced with the need to build a custom kernel at some point or another. The steady flow of kernel patches, device drivers, and new features that arrive on the scene is something of a mixed blessing. On one hand, it’s a privilege to live at the center of an active and vibrant software ecosystem. On the other hand, just keeping abreast of the constant flow of new material can be a job of its own.

If it ain’t broke, don’t fix it

Carefully weigh your site’s needs and risks when planning kernel upgrades and patches. A new release may be the latest and greatest, but is it as stable as the current version? Could the upgrade or patch be delayed and installed with another group of patches at the end of the month? Resist the temptation to let keeping up with the Joneses (in this case, the kernel-hacking community) dominate the best interests of your user community.

A good rule of thumb is to upgrade or apply patches only when the productivity gains you expect to obtain (usually measured in terms of reliability and performance) exceed the effort and lost time required for the installation. If you’re having trouble quantifying the specific gain, that’s a good sign that the patch can wait for another day. (Of course, security-related patches should be installed promptly.)

Setting up to build the Linux kernel

It’s less likely that you’ll need to build a kernel on your own if you’re running a distribution that uses a “stable” kernel to begin with. It used to be that the second part of the version number indicated whether the kernel was stable (even numbers) or in development (odd numbers). But these days, the kernel developers no longer follow that system. Check kernel.org for the official word on any particular kernel version. The kernel.org site is also the best source for Linux kernel source code if you are not relying on a particular distribution or vendor to provide you with a kernel.

Each distribution has a specific way to configure and build custom kernels. However, distributions also support the traditional way of doing things, which is what we describe here. It’s generally safest to use your distributor’s recommended procedure.

Configuring kernel options

Most distributions install kernel source files in versioned subdirectories under /usr/src/kernels. In all cases, you need to install the kernel source package before you can build a kernel on your system. See Chapter 6, Software Installation and Management, for tips on package installation.

Kernel configuration revolves around the .config file at the root of the kernel source directory. All the kernel configuration information is specified in this file, but its format is somewhat cryptic. Use the decoding guide in
to find out what the various options mean. It’s usually inadvisable to edit the .config file by hand because the effect of changing options is not always obvious. Options are frequently interdependent, so turning on an option might not be a simple matter of changing an n to a y.

To save folks from having to edit the .config file directly, Linux has several make targets that help you configure the kernel through a user interface. If you are running KDE, the prettiest configuration interface is provided by make xconfig. Likewise, if you’re running GNOME, make gconfig is probably the best option. These commands bring up a graphical configuration screen in which you can pick the devices to add to your kernel (or to compile as loadable modules).

If you are not running KDE or GNOME, you can use a terminal-based alternative invoked with make menuconfig. Finally, the bare-bones make config prompts you to respond to every single configuration option that’s available, which results in a lot of questions—and if you change your mind, you have to start over. We recommend make xconfig or make gconfig if your environment supports them; otherwise, use make menuconfig. Avoid make config, the least flexible and most painful option.

If you’re migrating an existing kernel configuration to a new kernel version (or tree), you can use the make oldconfig target to read in the previous config file and ask only the questions that are new to this edition of the kernel.

These tools are straightforward as far as the options you can turn on. Unfortunately, they are painful to use if you want to maintain multiple versions of the kernel to accompany multiple architectures or hardware configurations found in your environment.

All the various configuration interfaces described above generate a .config file that looks something like this:
As you can see, the contents are cryptic and do not attempt to describe what the various CONFIG tags mean. Each line refers to a specific kernel configuration option. The value `y` compiles the option into the kernel, and the value `m` enables the option as a loadable module.

Some options can be configured as modules and some can’t. You just have to know which is which; it will not be clear from the `.config` file. Nor are the CONFIG tags easily mapped to meaningful information.

The option hierarchy is extensive, so set aside many hours if you plan to scrutinize every possibility.

**Building the kernel binary**

Setting up an appropriate `.config` file is the most important part of the Linux kernel configuration process, but you must jump through several more hoops to turn that file into a finished kernel.

Here’s an outline of the entire process:

1. Change directory (`cd`) to the top level of the kernel source directory.
2. Run `make xconfig`, `make gconfig`, or `make menuconfig`.
3. Run `make clean`.
4. Run `make`.
5. Run `make modules_install`.
6. Run `make install`.
See [this page](#) for more information about GRUB.

You might also have to update, configure, and install the GRUB boot loader’s configuration file if this was not performed by the `make install` step. The GRUB updater scans the boot directory to see which kernels are available and automatically includes them in the boot menu.

The `make clean` step is not strictly necessary, but it’s generally a good idea to start with a clean build environment. In practice, many problems can be traced back to this step having been skipped.
Adding a Linux device driver

On Linux systems, device drivers are typically distributed in one of three forms:

- A patch against a specific kernel version
- A loadable kernel module
- An installation script or package that installs the driver

The most common form is the installation script or package. If you’re lucky enough to have one of these for your new device, you should be able to follow the standard procedure for installing new software.

In situations where you have a patch against a specific kernel version, you can in most cases install the patch with the following procedure:

```
linux# cd kernel_src_dir; patch -p1 < patch_file
```
FreeBSD supports the same three methods of changing kernel parameters as Linux: dynamically tuning the running kernel, building a new kernel from source, and loading dynamic modules.
Tuning FreeBSD kernel parameters

Many FreeBSD kernel parameters can be changed dynamically with the `sysctl` command, as is done on Linux. You can set values automatically at boot time by adding them to `/etc/sysctl.conf`. Many, many parameters can be changed this way; type `sysctl -a` to see them all. Not everything that shows up in the output of that command can be changed; many parameters are read-only.

The following paragraphs outline a few of the more commonly modified or interesting parameters that you might want to adjust.

- `net.inet.ip.forwarding` and `net.inet6.ip6.forwarding` control IP packet forwarding for IPv4 and IPv6, respectively.
- `kern.maxfiles` sets the maximum number of file descriptors that the system can open. You may need to increase this on systems such as database or web servers.
- `net.inet.tcp.mssdflt` sets the default TCP maximum segment size, which is the size of the TCP packet payload carried over IPv4. Certain payload sizes are too large for long-haul network links, and hence might be dropped by their routers. Changing this parameter can be useful when debugging long-haul connectivity issues.
- `net.inet.udp.blackhole` controls whether an ICMP “port unreachable” packet is returned when a packet arrives for a closed UDP port. Enabling this option (that is, disabling “port unreachable” packets) might slow down port scanners and potential attackers.
- `net.inet.tcp.blackhole` is similar in concept to the `udp.blackhole` parameter. TCP normally sends an RST (connection reset) response when packets arrive for closed ports. Setting this parameter to 1 prevents any SYN (connection setup) arriving on a closed port from generating an RST. Setting it to 2 prevents RST responses to any segment at all that arrives on a closed port.
- `kern.ipc.nmbclusters` controls the number of mbuf clusters available to the system. Mbufs are the internal storage structure for network packets, and mbuf clusters can be thought of as the mbuf “payload.” For servers that experience heavy network loads, this value may need to be increased from the default (currently 253,052 on FreeBSD 10).
- `kern.maxvnodes` sets the maximum number of vnodes, which are kernel data structures that track files. Increasing the number of available vnodes can improve disk throughput on a busy server. Examine the value of `vfs.numvnodes` on servers that are experiencing poor performance; if its value is close to the value of `kern.maxvnodes`, increase the latter.
Building a FreeBSD kernel

Kernel source comes from the FreeBSD servers in the form of a compressed tarball. Just download and unpack to install. Once the kernel source tree has been installed, the process for configuring and building the kernel is similar to that of Linux. However, the kernel source always lives in /usr/src/sys. Under that directory is a set of subdirectories, one for each architecture that is supported. Inside each architecture directory, a conf subdirectory includes a configuration file named GENERIC for the so-called “generic kernel,” which supports every possible device and option.

The configuration file is analogous to the Linux .config file. The first step in making a custom kernel is to copy the GENERIC file to a new, distinct name in the same directory, e.g., MYCUSTOM. The second step is to edit the config file and modify its parameters by commenting out functions and devices that you don’t need. The final step is to build and install the kernel. That final step must be performed in the top-level /usr/src directory.

FreeBSD kernel configuration files must be edited by hand. There are no dedicated user interfaces for this task as there are in the Linux world. Information on the general format is available from the config(5) man page, and information about how the config file is used can be found in the config(8) man page.

The configuration file contains some internal comments that describe what each option does. However, you do still need some background knowledge on a wide variety of technologies to make informed decisions about what to leave in. In general, you’ll want to leave all the options from the GENERIC configuration enabled and modify only the device-specific lines lower in the configuration file. It’s best to leave options enabled unless you’re absolutely certain you don’t need them.

For the final build step, FreeBSD has a single, highly automated make buildkernel target that combines parsing the configuration file, creating the build directories, copying the relevant source files, and compiling those files. This target accepts the custom configuration filename in the form of a build variable, KERNCONF. An analogous install target, make installkernel, installs the kernel and boot loader.

Here is a summary of the process:

1. Change directory (cd) to /usr/src/sys/arch/conf for your architecture.
2. Copy the generic configuration: cp GENERIC MYCUSTOM.
3. Edit your MYCUSTOM configuration file.
5. Run make buildkernel KERNCONF=MYCUSTOM.
6. Run `make installkernel KERNCONF=MYCUSTOM`.

Note that these steps are not cross-compilation-enabled! That is, if your build machine has an AMD64 architecture, you cannot `cd` to `/usr/src/sys/sparc/conf`, follow the normal steps, and end up with a SPARC-ready kernel.
11.6 Loadable kernel modules

Loadable kernel modules (LKMs) are available in both Linux and FreeBSD. LKM support allows a device driver—or any other kernel component—to be linked into and removed from the kernel while the kernel is running. This capability facilitates the installation of drivers because it avoids the need to update the kernel binary. It also allows the kernel to be smaller because drivers are not loaded unless they are needed.

Although loadable drivers are convenient, they are not 100% safe. Any time you load or unload a module, you risk causing a kernel panic. So don’t try out an untested module when you are not willing to crash the machine.

Like other aspects of device and driver management, the implementation of loadable modules is OS-dependent.
Loadable kernel modules in Linux

Under Linux, almost anything can be built as a loadable kernel module. The exceptions are the root filesystem type (whatever that might be on a given system) and the PS/2 mouse driver.

Loadable kernel modules are conventionally stored under /lib/modules/version, where version is the version of your Linux kernel as returned by `uname -r`.

You can inspect the currently loaded modules with the `lsmod` command:

```
redhat$ lsmod
Module Size Used by
ipmi_devintf 13064 2
ipmi_si 36648 1
ipmi_msghandler 31848 2 ipmi_devintf,ipmi_si
iptable_filter 6721 0
ip_tables 21441 1 iptable_filter
...
```

Loaded on this machine are the Intelligent Platform Management Interface modules and the `iptables` firewall, among other modules.

As an example of manually loading a kernel module, here’s how we would insert a module that implements sound output to USB devices:

```
redhat$ sudo modprobe snd-usb-audio
```

We can also pass parameters to modules as they are loaded; for example,

```
redhat$ sudo modprobe snd-usb-audio nrpacks=8 async_unlink=1
```

`modprobe` is a semi-automatic wrapper around a more primitive command, `insmod`. `modprobe` understands dependencies, options, and installation and removal procedures. It also checks the version number of the running kernel and selects an appropriate version of the module from within `/lib/modules`. It consults the file `/etc/modprobe.conf` to figure out how to handle each individual module.

Once a loadable kernel module has been manually inserted into the kernel, it remains active until you explicitly request its removal or reboot the system. You could use `modprobe -r snd-usb-audio` to remove the audio module loaded above. Removal works only if the number of current references to the module (listed in the “Used by” column of `lsmod`’s output) is 0.

You can dynamically generate an `/etc/modprobe.conf` file that corresponds to all your currently installed modules by running `modprobe -c`. This command generates a long file that looks like this:
The *path* statements tell where a particular module can be found. You can modify or add entries of this type to keep your modules in a nonstandard location.

The *alias* statements map between module names and block-major device numbers, character-major device numbers, filesystems, network devices, and network protocols.

The *options* lines are not dynamically generated but must be manually added by an administrator. They specify options that should be passed to a module when it is loaded. For example, you could use the following line to pass in additional options to the USB sound module:

```
options snd-usb-audio nrpacks=8 async_unlink=1
```

*modprobe* also understands the statements *install* and *remove*. These statements allow commands to be executed when a specific module is inserted into or removed from the running kernel.
Loadable kernel modules in FreeBSD

Kernel modules in FreeBSD live in /boot/kernel (for standard modules that are part of the distribution) or /boot/modules (for ported, proprietary, and custom modules). Each kernel module uses the .ko filename extension, but it is not necessary to specify that extension when loading, unloading, or viewing the status of the module.

For example, to load a module named foo.ko, run kldload foo in the appropriate directory. To unload the module, run kldunload foo from any location. To view the module’s status, run kldstat -m foo from any location. Running kldstat without any parameters displays the status of all currently loaded modules.

Modules listed in either of the files /boot/defaults/loader.conf (system defaults) or /boot/loader.conf are loaded automatically at boot time. To add a new entry to /boot/loader.conf, use a line of the form

```
zfs_load=YES
```

The appropriate variable name is just the module basename with _load appended to it. The line above ensures that the module /boot/kernel/zfs.ko will be loaded at boot; it implements the ZFS filesystem.
11.7 Booting

Now that we have covered kernel basics, it’s time to learn what actually happens when a kernel loads and initializes at startup. You’ve no doubt seen countless boot messages, but do you know what all of those messages actually mean?

The following messages and annotations come from some key phases of the boot process. They almost certainly won’t be an exact match for what you see on your own systems and kernels. However, they should give you a notion of some of the major themes in booting and a feeling for how the Linux and FreeBSD kernels start up.
Linux boot messages

The first boot log we examine is from a CentOS 7 machine running a 3.10.0 kernel.

Feb 14 17:18:57 localhost kernel: Initializing cgroup subsys cpuset
Feb 14 17:18:57 localhost kernel: Initializing cgroup subsys cpu
Feb 14 17:18:57 localhost kernel: Initializing cgroup subsys cpuacct
Feb 14 17:18:57 localhost kernel: Linux version 3.10.0-327.el7.x86_64
(builder@kbuilder.dev.centos.org) (gcc version 4.8.3 20140911 (Red
Hat 4.8.3-9) (GCC) ) #1 SMP Thu Nov 19 22:10:57 UTC 2015
Feb 14 17:18:57 localhost kernel: Command line: BOOT_IMAGE=/
vmlinuz-3.10.0-327.el7.x86_64 root=/dev mapper/centos-root ro
  crashkernel=auto rd.lvm.lv=centos/root rd.lvm.lv=centos/swap rhgb
  quiet LANG=en_US.UTF-8

These initial messages tell us that the top-level control groups (cgroups) are starting up on a Linux 3.10.0 kernel. The messages tell us who built the kernel and where, and which compiler they used (gcc). Note that although this log comes from a CentOS system, CentOS is a clone of Red Hat, and the boot messages remind us of that fact.

The parameters set in the GRUB boot configuration and passed from there into the kernel are listed above as the “command line.”

Feb 14 17:18:57 localhost kernel: e820: BIOS-provided physical RAM map:
Feb 14 17:18:57 localhost kernel: BIOS–e820: [mem
  0x0000000000000000–0x0000000000009bff] usable
Feb 14 17:18:57 localhost kernel: BIOS–e820: [mem 0x00000000000009fc00–
  0x0000000000009fff] reserved
  ...
Feb 14 17:18:57 localhost kernel: Hypervisor detected: KVM
Feb 14 17:18:57 localhost kernel: AGP: No AGP bridge found
Feb 14 17:18:57 localhost kernel: x86 PAT enabled: cpu 0, old
  0x7040f0007040f00, new 0x7010600070106
Feb 14 17:18:57 localhost kernel: CPU MTRRs all blank – virtualized
  system.
Feb 14 17:18:57 localhost kernel: e820: last_pfn = 0xffff0 max_arch_pfn
  = 0x40000000
Feb 14 17:18:57 localhost kernel: found SMP MP-table at [mem
  0x0009fff0–0x0009ffff] mapped at [ffff8000009fff0]
Feb 14 17:18:57 localhost kernel: init_memory_mapping: [mem
  0x00000000–0x000fffff]
  ...

These messages describe the processor that the kernel has detected and show how the RAM is mapped. Note that the kernel is aware that it’s booting within a hypervisor and is not actually running on bare hardware.
Here the kernel initializes the system’s various data buses, including the PCI bus and the USB subsystem.

These messages document the kernel’s discovery of various devices, including the power button, a USB hub, a mouse, and a real-time clock (RTC) chip. Some of the “devices” are metadevices.
rather than actual hardware; these constructs manage groups of real, related hardware devices. For example, the usbhid (USB Human Interface Device) driver manages keyboards, mice, tablets, game controllers, and other types of input devices that follow USB reporting standards.

In this phase, the kernel initializes a variety of network drivers and facilities.

The drop monitor is a Red Hat kernel subsystem that implements comprehensive monitoring of network packet loss. “TCP cubic” is a congestion-control algorithm optimized for high-latency, high-bandwidth connections, so-called long fat pipes.

As mentioned [here](#), Netlink sockets are a modern approach to communication between the kernel and user-level processes. The XFRM Netlink socket is the link between the user-level IPsec process and the kernel’s IPsec routines.

The last two lines document the registration of two additional network protocol families.

Like other OSs, CentOS provides a way to incorporate and validate updates. The validation portion uses X.509 certificates that are installed into the kernel.

Here, the kernel reports that it’s unable to find a Trusted Platform Module (TPM) on the system. TPM chips are cryptographic hardware devices that provide for secure signing operations. When used properly, they can make it much more difficult to hack into a system.

For example, the TPM can be used to sign kernel code and to make the system refuse to execute any portion of the code for which the current signature doesn’t match the TPM signature. This
measure helps avoid the execution of maliciously injected code. An admin who expects to have a working TPM would be unhappy to see this message!

The last message shows the kernel setting the battery-backed real-time clock to the current time of day. This is the same RTC that we saw mentioned earlier when it was identified as a device.

Feb 14 17:18:57 localhost kernel: e1000: Intel(R) PRO/1000 Network Driver – version 7.3.21-k8-NAPI
Feb 14 17:18:57 localhost kernel: e1000: Copyright (c) 1999-2006 Intel Corporation.
Feb 14 17:18:58 localhost kernel: e1000 0000:00:03.0 eth0: (PCI:33MHz:32-bit) 08:00:27:d0:ae:6f
Feb 14 17:18:58 localhost kernel: e1000 0000:00:03.0 eth0: Intel(R) PRO/1000 Network Connection

See this page for more information about DHCP.

Now the kernel has found the gigabit Ethernet interface and initialized it. The interface’s MAC address (08:00:27:d0:ae:6f) might be of interest to you if you wanted this machine to obtain its IP address through DHCP. Specific IP addresses are often locked to specific MACs in the DHCP server configuration so that servers can have IP address continuity.
Here the kernel recognizes and initializes various drives and support devices (hard disk drives, a SCSI-based virtual CD-ROM, and an ATA hard disk). It also mounts a filesystem (XFS) that is part of the device-mapper subsystem (the dm-0 filesystem).

As you can see, Linux kernel boot messages are verbose almost to a fault. However, you can rest assured that you’ll see everything the kernel is doing as it starts up, a most useful feature if you encounter problems.
FreeBSD boot messages

The log below is from a FreeBSD 10.3-RELEASE system that runs the kernel shipped with the release. Much of the output will look eerily familiar; the sequence of events is quite similar to that found in Linux. One notable difference is that the FreeBSD kernel produces far fewer boot messages than does Linux. Compared to Linux, FreeBSD is downright taciturn.

Sep 25 12:48:36 bucephalus kernel: FreeBSD 10.3-RELEASE #0
r297264: Fri Mar 25 02:10:02 UTC 2016
Sep 25 12:48:36 bucephalus kernel: root@releng1.ny.freebsd.org:
/usr/obj/usr/src/sys GENERIC amd64
Sep 25 12:48:36 bucephalus kernel: FreeBSD clang version 3.4.1
(tags/RELEASE_34/dot1-final 208032) 20140512

The initial messages above tell you the OS release, the time at which the kernel was built from source, the name of the builder, the configuration file that was used, and finally, the compiler that generated the code (Clang version 3.4.1: a compiler front end, really; but let’s not quibble).

Sep 25 12:48:36 bucephalus kernel: real memory = 4831838208 (4608 MB)
Sep 25 12:48:36 bucephalus kernel: avail memory = 4116848640 (3926 MB)

Above are the system’s total amount of memory and the amount that’s available to user-space code. The remainder of the memory is reserved for the kernel itself.

Total memory of 4608MB probably looks a bit strange. However, this FreeBSD instance is running under a hypervisor. The amount of “real memory” is an arbitrary value that was set when the virtual machine was configured. On bare-metal systems, the total memory is likely to be a power of 2, since that’s how actual RAM chips are manufactured (e.g., 8192MB).

Sep 25 12:48:36 bucephalus kernel: vgapci0: <VGA-compatible display>
mem 0xe00000000-0xe0fffff irq 18 at device 2.0 on pc10
Sep 25 12:48:36 bucephalus kernel: vgapci0: Boot video device

There’s the default video display, which was found on the PCI bus. The output shows the memory range to which the frame buffer has been mapped.

Sep 25 12:48:36 bucephalus kernel: em0: <Intel(R) PRO/1000 Legacy Network Connection 1.1.0> port 0xd010-0xd017 mem 0xf0000000-
0xf01fff f irq 19 at device 3.0 on pc10
Sep 25 12:48:36 bucephalus kernel: em0: Ethernet address:
08:00:27:b5:49:fc

And above, the Ethernet interface, along with its hardware (MAC) address.
As shown above, the kernel initializes the USB bus, the USB hub, the CD-ROM drive (actually a DVD-ROM drive, but virtualized to look like a CD-ROM), and the ada disk driver.

The final messages in the FreeBSD boot log document a variety of odds and ends. The “random” pseudo-device harvests entropy from the system and generates random numbers. The kernel seeded its number generator and put it in nonblocking mode. A few other devices came up, and the kernel mounted the root filesystem.

At this point, the kernel boot messages end. Once the root filesystem has been mounted, the kernel transitions to multiuser mode and initiates the user-level startup scripts. Those scripts, in turn, start the system services and make the system available for use.
11.8 Booting alternate kernels in the cloud

Cloud instances boot differently from traditional hardware. Most cloud providers sidestep GRUB and use either a modified open source boot loader or some kind of scheme that avoids the use of a boot loader altogether. Therefore, booting an alternate kernel on a cloud instance usually requires that you interact with the cloud provider’s web console or API.

This section briefly outlines some of the specifics that relate to booting and kernel selection on our example cloud platforms. For a more general introduction to cloud systems, see Chapter 9, Cloud Computing.

On AWS, you’ll need to start with a base AMI (Amazon machine image) that uses a boot loader called PV-GRUB. PV-GRUB runs a patched version of legacy GRUB and lets you specify the kernel in your AMI’s menu.lst file.

After compiling a new kernel, edit /boot/grub/menu.lst to add it to the boot list:

```
default 0
fallback 1
timeout 0
hiddenmenu

title My Linux Kernel
root (hd0)
kernel /boot/my-vmlinuz-4.3 root=LABEL=/ console=hvc0
initrd /boot/my-initrd.img-4.3

title Amazon Linux
root (hd0)
kernel /boot/vmlinuz-4.1.10-17.31.amzn1.x86 root=LABEL=/ console=hvc0
initrd /boot/initramfs-4.1.10-17.31.amzn1.x86.img
```

Here, the custom kernel is the default, and the fallback option points to the standard Amazon Linux kernel. Having a fallback helps ensure that your system can boot even if your custom kernel can’t be loaded or doesn’t work correctly. See Amazon EC2’s User Guide for Linux Instances for more details on this process.

Historically, DigitalOcean bypassed the boot loader through a QEMU (short for Quick Emulator) feature that allowed a kernel and RAM disk to be loaded directly into a droplet. Thankfully, DigitalOcean now allows droplets to use their own boot loaders. Most modern operating systems are supported, including CoreOS, FreeBSD, Fedora, Ubuntu, Debian, and CentOS. Changes to boot options, including selection of the kernel, are handled by the respective OS boot loaders (GRUB, usually).

Google Cloud Platform (GCP) is the most flexible platform when it comes to boot management. Google lets you upload complete system disk images to your Compute Engine account. Note that
in order for a GCP image to boot properly, it must use the MBR partitioning scheme and include an appropriate (installed) boot loader. UEFI and GPT do not apply here!

The cloud.google.com/compute/docs/creating-custom-image tutorial on building images is incredibly thorough and specifies not only the required kernel options but also the recommended settings for kernel security.
11.9 Kernel errors

Kernel crashes (aka kernel panics) are an unfortunate reality that can happen even on properly configured systems. They have a variety of causes. Bad commands entered by privileged users can certainly crash the system, but a more common cause is faulty hardware. Physical memory failures and hard drive errors (bad sectors on a platter or device) are both notorious for causing kernel panics.

It’s also possible for bugs in the implementation of the kernel to result in crashes. However, such crashes are exceedingly rare in kernels anointed as “stable.” Device drivers are another matter, however. They come from many different sources and are often of less-than-exemplary code quality.

If hardware is the underlying cause of a crash, keep in mind that the crash may have occurred long after the device failure that sparked it. For example, you can often remove a hot-swappable hard drive without causing immediate problems. The system continues to hum along without (much) complaint until you try to reboot or perform some other operation that depends on that particular drive.

Despite the names “panic” and “crash,” kernel panics are usually relatively structured events. User-space programs rely on the kernel to police them for many kinds of misbehavior, but the kernel has to monitor itself. Consequently, kernels include a liberal helping of sanity-checking code that attempts to validate important data structures and invariants in passing. None of those checks should ever fail; if they do, it’s sufficient reason to panic and halt the system, and the kernel does so proactively.

Or at least, that’s the traditional approach. Linux has liberalized this rule somewhat through the “oops” system; see the next section.
**Linux kernel errors**

Linux has four varieties of kernel failure: soft lockups, hard lockups, panics, and the infamous Linux “oops.” Each one of these usually provides a complete stack trace, except for certain soft lockups that are recoverable without a panic.

A soft lockup occurs when the system is in kernel mode for more than a few seconds, thus preventing user-level tasks from running. The interval is configurable, but it is usually around 10 seconds, which is a long time for a process to be denied CPU cycles! During a soft lockup, the kernel is the only thing running, but it is still servicing interrupts such as those from network interfaces and keyboards. Data is still flowing in and out of the system, albeit in a potentially crippled fashion.

A hard lockup is the same as a soft lockup, but with the additional complication that most processor interrupts go unserviced. Hard lockups are overtly pathological conditions that are detected relatively quickly, whereas soft lockups can occur even on correctly configured systems that are experiencing some kind of extreme condition, such as a high CPU load.

In both cases, a stack trace and a display of the CPU registers (a “tombstone”) are usually dumped to the console. The trace shows the sequence of function calls that resulted in the lockup. In most cases, this trace tells you quite a bit about the cause of the problem.

A soft or hard lockup is almost always the result of a hardware failure, the most common culprit being bad memory. The second most common reason for a soft lockup is a kernel spinlock that has been held too long; however, this situation normally occurs only with nonstandard kernel modules. If you are running any unusual modules, try unloading them and see if the problem recurs.

When a lockup occurs, the usual behavior is for the system to stay frozen so that the tombstone remains visible on the console. But in some environments, it’s preferable to have the system panic and thus reboot. For example, an automated test rig needs systems to avoid hanging, so these systems are often configured to reboot into a safe kernel after encountering a lockup.

`sysctl` can configure both soft and hard lockups to panic:

```
linux$ sudo sysctl kernel.softlockup_panic=1
linux$ sudo sysctl kernel.nmi_watchdog=1
```

You can set these parameters at boot by listing them in `/etc/sysctl.conf`, just like any other kernel parameter.

The Linux “oops” system is a generalization of the traditional UNIX “panic after any anomaly” approach to kernel integrity. Oops doesn’t stand for anything; it’s just the English word oops, as in “Oops! I zeroed out your SAN again.” Oopses in the Linux kernel can lead to a panic, but they needn’t always. If the kernel can repair or address an anomaly through a less drastic measure, such as killing an individual process, it might do that instead.
When an oops occurs, the kernel generates a tombstone in the kernel message buffer that’s viewable with the `dmesg` command. The cause of the oops is listed at the top. It might be something like “unable to handle kernel paging request at virtual address 0x0000000000000.”

You probably won’t be debugging your own kernel oopses. However, your chance of attracting the interest of a kernel or module developer is greatly increased if you do a good job of capturing the available context and diagnostic information, including the full tombstone.

The most valuable information is at the beginning of the tombstone. That fact can present a problem after a full-scale kernel panic. On a physical system, you may be able to just go to the console and page up through the history to see the full dump. But on a virtual machine, the console might be a window that becomes frozen when the Linux instance panics; it depends on the hypervisor. If the text of the tombstone has scrolled out of view, you won’t be able to see the cause of the crash.

One way to minimize the likelihood of information loss is to increase the resolution of the console screen. We’ve found that a resolution of 1280 × 1024 is adequate to display the full text of most kernel panics.

You can set the console resolution by modifying `/etc/grub2/grub.cfg` and adding `vga=795` as a kernel startup parameter for the kernel you want to boot. You can also set the resolution by adding this clause to the kernel “command line” from GRUB’s boot menu screen. The latter approach lets you test the waters without making any permanent changes.

To make the change permanent, find the menu item with the boot command for the kernel that you wish to boot, and modify it. For example, if the boot command looks like this:

```bash
linux16 /vmlinuz-3.10.0-229.el7.x86_64 root=/devmapper/centos-root ro rd.lvm.lv=centos/root rd.lvm.lv=centos/swap crashkernel=auto biosdevname=0 net.ifnames=0 LANG=en_US.UTF-8
```

then simply modify it to add the `vga=795` parameter at the end:

```bash
linux16 /vmlinuz-3.10.0-229.el7.x86_64 root=/devmapper/centos-root ro rd.lvm.lv=centos/root rd.lvm.lv=centos/swap crashkernel=auto biosdevname=0 net.ifnames=0 LANG=en_US.UTF-8 vga=795
```

Other resolutions can be achieved by setting the `vga` boot parameter to other values. Table 11.6 lists the possibilities.

Table 11.6: VGA mode values
<table>
<thead>
<tr>
<th>Geometry</th>
<th>8</th>
<th>15</th>
<th>16</th>
<th>24</th>
</tr>
</thead>
<tbody>
<tr>
<td>640 x 480</td>
<td>769</td>
<td>784</td>
<td>785</td>
<td>786</td>
</tr>
<tr>
<td>800 x 600</td>
<td>771</td>
<td>787</td>
<td>788</td>
<td>789</td>
</tr>
<tr>
<td>1024 x 768</td>
<td>773</td>
<td>790</td>
<td>791</td>
<td>792</td>
</tr>
<tr>
<td>1280 x 1024</td>
<td>775</td>
<td>793</td>
<td>884</td>
<td>795</td>
</tr>
<tr>
<td>1400 x 1050</td>
<td>834</td>
<td>834</td>
<td>834</td>
<td>834</td>
</tr>
<tr>
<td>1600 x 1200</td>
<td>884</td>
<td>884</td>
<td>884</td>
<td>884</td>
</tr>
</tbody>
</table>
FreeBSD kernel panics

FreeBSD does not divulge much information when the kernel panics. If you are running a generic kernel from a production release, the best thing to do if you are encountering regular panics is to instrument the kernel for debugging. Rebuild the generic kernel with `makeoptions DEBUG=-g` enabled in the kernel configuration, and reboot with that new kernel. Once the system panics again, you can use `kgdb` to generate a stack trace from the resulting crash dump in `/var/crash`.

Of course, if you’re running unusual kernel modules and the kernel doesn’t panic when you don’t load them, that’s a good indication of where the issue lies.

An important note: crash dumps are the same size as real (physical) memory, so you must ensure that `/var/crash` has at least that much space available before you enable these dumps. There are ways to get around this, though: for more information, see the man pages for `dumpon` and `savecore` and the `dumpdev` variable in `/etc/rc.conf`. 
11.10 Recommended Reading

You can visit lwn.net for the latest information on what the kernel community is doing. In addition, we recommend the following books.


12 Printing

Printing is a necessary evil. No one wants to deal with it, but every user wants to print. For better or worse, printing on UNIX and Linux systems typically requires at least some configuration and occasionally some coddling by a system administrator.

Ages ago, there were three common printing systems: BSD, System V, and CUPS (the Common UNIX Printing System). Today, Linux and FreeBSD both use CUPS, an up-to-date, sophisticated, network- and security-aware printing system. CUPS includes a modern, browser-based GUI as well as shell-level commands that allow the printing system to be controlled by scripts.

Before we start, a general point: system administrators often consider printing a lower priority than do users. Administrators are accustomed to reading documents on-line, but users often need hard copy, and they want the printing system to work 100% of the time. Satisfying these desires is one of the easier ways for sysadmins to earn Brownie points with users.

Printing relies on a handful of pieces:

- A print “spooler” that collects and schedules jobs. The word “spool” originated as an acronym for Simultaneous Peripheral Operation On-Line. Now it’s just a generic term.
- User-level utilities (command-line interfaces or GUIs) that talk to the spooler. These utilities send jobs to the spooler, query the system about jobs (both pending and complete), remove or reschedule jobs, and configure the other parts of the system.

- Back ends that talk to the printing devices themselves. (These are normally unseen and hidden under the floorboards.)

- A network protocol that lets spoolers communicate and transfer jobs.

Modern environments often use network-attached printers that minimize the amount of setup and processing that must be done on the UNIX or Linux side.
12.1 CUPS PRINTING

CUPS was created by Michael Sweet and has been adopted as the default printing system for Linux, FreeBSD, and macOS. Michael has been at Apple since 2007, where he continues to develop CUPS and its ecosystem.

Just as newer mail transport systems include a command called `sendmail` that lets older scripts (and older system administrators!) work as they always did back in `sendmail`’s glory days, CUPS supplies traditional commands such as `lp` and `lpr` that are backward compatible with legacy UNIX printing systems.

CUPS servers are also web servers, and CUPS clients are web clients. The clients can be commands such as the CUPS versions of `lpr` and `lpq`, or they can be applications with their own GUIs. Under the covers they’re all web apps, even if they’re talking only to the CUPS daemon on the local system. CUPS servers can also act as clients of other CUPS servers.

A CUPS server offers a web interface to its full functionality on port 631. For administrators, a web browser is usually the most convenient way to manage the system; just navigate to `http://printhost:631`. If you need secure communication with the daemon (and your system offers it), use `https://printhost:433` instead. Scripts can use discrete commands to control the system, and users normally access it through a GNOME or KDE interface. These routes are all equivalent.

HTTP is the underlying protocol for all interactions among CUPS servers and their clients. Actually, it’s the Internet Printing Protocol, a souped-up version of HTTP. Clients submit jobs with the HTTP/IPP POST operation and request status with HTTP/IPP GET. The CUPS configuration files also look suspiciously similar to Apache web server configuration files.
Interfaces to the printing system

CUPS printing is often done from a GUI, and administration is often done through a web browser. As a sysadmin, though, you (and perhaps some of your hard-core terminal users) might want to use shell-level commands as well. CUPS includes work-alike commands for many of the basic, shell-level printing commands of the legacy BSD and System V printing systems. Unfortunately, CUPS doesn’t necessarily emulate all the bells and whistles. Sometimes, it emulates the old interfaces entirely too well; instead of giving you a quick usage summary, `lpr --help` and `lp --help` just print error messages.

Here’s how you might print the files `foo.pdf` and `/tmp/testprint.ps` to your default printer under CUPS:

```
$ lpr foo.pdf /tmp/testprint.ps
```

The `lpr` command transmits copies of the files to the CUPS server, `cupsd`, which stores them in the print queue. CUPS processes each file in turn as the printer becomes available.

When printing, CUPS examines both the document and the printer’s PostScript Printer Description (PPD) file to see what needs to be done to get the document to print properly. (Despite the name, PPD files are used even for non-PostScript printers.)

To prepare a job for printing on a specific printer, CUPS passes it through a series of filters. For example, one filter might reformat the job so that two reduced-size page images print on each physical page (aka “2-up printing”), and another might transform the job from PostScript to PCL. Filters can also do printer-specific processing such as printer initialization. Some filters perform rasterization, turning abstract instructions such as “draw a line across the page” into a bitmap image. Such rasterizers are useful for printers that do not include their own rasterizers or that don’t speak the language in which a job was originally submitted.

The final stage of the print pipeline is a back end that transmits the job from the host to the printer through an appropriate protocol such as Ethernet. The back end also communicates status information in the other direction, back to the CUPS server. After transmitting the print job, the CUPS daemon returns to processing its queues and handling requests from clients, and the printer goes off to print the job it was shipped.
The print queue

cupsd’s centralized control of the printing system makes it easy to understand what the user-level commands are doing. For example, the `lpq` command requests job status information from the server and reformats it for display. Other CUPS clients ask the server to suspend, cancel, or reprioritize jobs. They can also move jobs from one queue to another.

Most changes require jobs to be identified by their job number, which you can get from `lpq`. For example, to remove a print job, just run `lprm jobid`.

`lpstat -t` summarizes the print server’s overall status.
Multiple printers and queues

The CUPS server maintains a separate queue for each printer. Command-line clients accept an option (typically `-P printer` or `-p printer`) by which you specify the queue to address. You can also set a default printer for yourself by setting the PRINTER environment variable.

```bash
$ export PRINTER=printer_name
```

or by telling CUPS to use a particular default for your account.

```bash
$ lpoptions -d printer_name
```

When run as root, `lpoptions` sets system-wide defaults in `/etc/cups/lpoptions`, but it’s more typically used by individual, nonroot users. `lpoptions` lets each user define personal printer instances and defaults, which it stores in `~/.cups/lpoptions`. The command `lpoptions -l` lists the current settings.
Printer instances

If you have only one printer but want to use it in several ways—say, both for quick drafts and for final production work—CUPS lets you set up different “printer instances” for these different uses.

For example, if you already have a printer named Phaser_6120, the command

```bash
$ loptions -p Phaser_6120/2up -o number-up=2 -o job-sheets=standard
```

creates an instance named Phaser_6120/2up that performs 2-up printing and adds banner pages. Once the instance has been created, the command

```bash
$ lpr -P Phaser_6120/2up biglisting.ps
```

prints the PostScript file `biglisting.ps` as a 2-up job with a banner page.
Network printer browsing

From CUPS’s perspective, a network of machines isn’t very different from an isolated machine. Every computer runs a `cupsd`, and all the CUPS daemons talk to one another.

If you’re working on the command line, you configure a CUPS daemon to accept print jobs from remote systems by editing the `/etc/cups/cupsd.conf` file (see [this page](#)). By default, servers that are set up this way broadcast information every 30 seconds about the printers they serve. As a result, computers on the local network automatically learn about the printers that are available to them. You can effect the same configuration by clicking a checkbox in the CUPS GUI in your browser.

If someone has plugged in a new printer, if you’ve brought your laptop into work, or if you’ve just installed a new workstation, you can tell `cupsd` to redetermine what printing services are available; click the Find New Printers button in the Administration tab of the CUPS GUI.

Because broadcast packets do not cross subnet boundaries, it’s a bit trickier to make printers available to multiple subnets. One solution is to designate, on each subnet, a slave server that polls the other subnets’ servers for information and then relays that information to machines on the local subnet.

For example, suppose the print servers allie (192.168.1.5) and jj (192.168.2.14) live on different subnets and we want both of them to be accessible to users on a third subnet, 192.168.3. We designate a slave server (say, copeland, 192.168.3.10) and add these lines to its `cupsd.conf` file:

```
BrowsePoll allie
BrowsePoll jj
BrowseRelay 127.0.0.1 192.168.3.255
```

The first two lines tell the slave’s `cupsd` to poll the `cupsd` on allie and jj for information about the printers they serve. The third line tells copeland to relay the information it learns to its own subnet. Simple!
Filters

Rather than using a specialized printing tool for every printer, CUPS uses a chain of filters to convert each printed file into a form the destination printer can understand.

The CUPS filter scheme is elegant. Given a document and a target printer, CUPS uses its .types files to figure out the document’s MIME type. It consults the printer’s PPD file to figure out what MIME types the printer can handle. It then uses .convs files to deduce what filter chains could convert one format to the other, and what each prospective chain would cost. Finally, it picks a chain and passes the document through those filters. The final filter in the chain passes the printable format to a back end, which transmits the data to the printer through whatever hardware or protocol the printer understands.

We can flesh out that process a bit. CUPS uses rules in /usr/share/cups/mime/mime.types to figure out the incoming data type. For example, the rule

```
application/pdf           pdf string (0,%PDF)
```

means “If a file has a .pdf extension or starts with the string %PDF, then its MIME type is application/pdf.”

CUPS figures out how to convert one data type to another by looking up rules in the file mime.convs (usually in /etc/cups or /usr/share/cups/mime). For example,

```
application/pdf           application/postscript 33 pdftops
```

means “To convert an application/pdf file to an application/postscript file, run the filter pdftops.” The number 33 is the cost of the conversion. When CUPS finds that several filter chains can convert a file from one type to another, it picks the chain with the lowest total cost. (Costs are chosen by whoever creates the mime.convs file—the distribution maintainers, perhaps. If you want to spend time tuning them because you think you can do a better job, you may have too much free time.)

The last component in a CUPS pipeline is a filter that talks directly to the printer. In the PPD of a non-PostScript printer, you might see lines such as

```
*cupsFilter: "application/vnd.cups-postscript @ foomatic-rip"
```

or even

```
*cupsFilter: "application/vnd.cups-postscript foomatic-rip"
```

The quoted string has the same format as a line in mime.convs, but there’s only one MIME type instead of two. This line advertises that the foomatic-rip filter converts data of type application/vnd.cups-postscript to the printer’s native data format. The cost is zero (or omitted)
because there’s only one way to do this step, so why pretend there’s a cost? (Some PPDs for non-
PostScript printers, like those from the Gutenprint project, are slightly different.)

To find the filters available on your system, try running `locate pstops`. `pstops` is a popular filter
that massages PostScript jobs in various ways, such as adding a PostScript command to set the
number of copies. Wherever you find `pstops`, the other filters won’t be far away.

You can ask CUPS for a list of the available back ends by running `lpinfo -v`. If your system lacks
a back end for the network protocol you need, it may be available from the web or from your
Linux distributor.
12.2 CUPS SERVER ADMINISTRATION

**cupsd** starts at boot time and runs continuously. All our example systems are set up this way by default.

> See [this page](#) for details about Apache configuration.

The CUPS configuration file, **cupsd.conf**, is usually found in `/etc/cups`. The file format is similar to that of the Apache configuration file. If you’re comfortable with one of these files, you’ll be comfortable with the other. You can view and edit **cupsd.conf** with a text editor or, once again, from the CUPS web GUI.

The default config file is well commented. The comments and the **cupsd.conf** man page are good enough that we won’t belabor the details of configuration here.

CUPS reads its configuration file only at startup. If you change the contents of **cupsd.conf**, you must restart **cupsd** for changes to take effect. If you make changes through **cupsd**’s web GUI, **cupsd** restarts automatically.
Network print server setup

If you’re having trouble printing over the network, review the browser-based CUPS GUI and make sure you’ve checked all the right boxes. Possible problem areas include an unpublished printer, a CUPS server that isn’t broadcasting its printers to the network, or a CUPS server that won’t accept network print jobs.

If you’re editing the cupsd.conf file directly, you’ll need to make a couple of changes. First, change

```
<Location />
Order Deny,Allow
Deny From All
Allow From 127.0.0.1
</Location>
```

to

```
<Location />
Order Deny,Allow
Deny From All
Allow From 127.0.0.1
Allow From netaddress
</Location>
```

Replace netaddress with the IP address of the network from which you want to accept jobs (e.g., 192.168.0.0).

Then, look for the BrowseAddress keyword and set it to the broadcast address on that network plus the CUPS port; for example,

```
BrowseAddress 192.168.0.255:631
```

These steps tell the server to accept requests from any machine on the designated subnet and to broadcast what it knows about the printers it’s serving to every CUPS daemon on that network. That’s it! Once you restart cupsd, it comes back as a server.
**Printer autoconfiguration**

You can use CUPS without a printer (e.g., to convert files to PDF or fax format), but its typical role is to manage real printers. In this section, we review the ways in which you can deal with the printers per se.

In some cases, adding a printer is trivial. CUPS autodetects USB printers when they’re plugged into the system and figures out what to do with them.

Even if you have to do some configuration work yourself, adding a printer is often no more painful than plugging in the hardware, connecting to the CUPS web interface at localhost:631/admin, and answering a few questions. KDE and GNOME come with their own printer configuration widgets, which you may prefer to the CUPS interface. (We like the CUPS GUI.)

If someone else adds a printer and one or more CUPS servers running on the network know about it, your CUPS server will learn of its existence. You don’t have to explicitly add the printer to the local inventory or copy PPDs to your machine. It’s magic.
Network printer configuration

Network printers—that is, printers whose primary hardware interface is an Ethernet jack or Wi-Fi radio—need some configuration of their own just to be proper citizens of the TCP/IP network. In particular, they need to know their own IP addresses and netmasks. That information is usually conveyed to them in one of two ways.

Network printers can get this information from a BOOTP or DHCP server, and this method works well in environments that have many such printers. See DHCP: the Dynamic Host Configuration Protocol for more information about DHCP.

Alternatively, you can assign the printer a static IP address from its console, which usually consists of a set of buttons on the printer’s front panel and a one-line display. Fumble around with the menus until you discover where to set the IP address. (If there is a menu option to print the menus, use it and put the printed version underneath the printer for future reference.)

Once configured, network printers usually have a web console that’s accessible from a browser. However, printers must have an IP address and must be up and running on the network before you can access them this way, so this interface is unavailable just when it’s most needed.

DHCP: the Dynamic Host Configuration Protocol
Printer configuration examples

Below, we add the parallel printer groucho and the network printer fezmo from the command line:

```
$ sudo lpadmin -p groucho -E -v parallel:/dev/lp0 -m pxlcolor.ppd
$ sudo lpadmin -p fezmo -E -v socket://192.168.0.12 -m laserjet.ppd
```

Groucho is attached to port /dev/lp0 and fezmo is at IP address 192.168.0.12. We specify each device in the form of a universal resource indicator (URI) and choose an appropriate PPD from the ones in /usr/share/cups/model.

As long as cupsd has been configured as a network server, it immediately makes the new printers available to other clients on the network. No restart is required.

CUPS accepts a wide variety of URIs for printers. Here are a few more examples:

- ipp://zoe.admin.com/ipp
- lpd://riley.admin.com/ps
- serial://dev/ttyS0?baud=9600+parity=even+bits=7
- socket://gillian.admin.com:9100
- usb://XEROX/Phaser%206120?serial=YGG210547

Some types take options (e.g., serial) and others don’t. lpinfo -v lists the devices your system can see and the types of URIs that CUPS understands.
Service shutoff

Removing a printer is easily done with `lpadmin -x`:

```
$ sudo lpadmin -x fezmo
```

OK, but what if you just want to disable a printer temporarily for service instead of removing it? You can block the print queue at either end. If you disable the tail (the exit or printer side) of the queue, users can still submit jobs, but the jobs won’t print until the outlet is re-enabled. If you disable the head (the entrance) of the queue, jobs that are already in the queue can still print, but the queue rejects attempts to submit new jobs.

The `cupsdisable` and `cupsenable` commands control the exit side of the queue, and the `reject` and `accept` commands control the submission side. For example,

```
$ sudo cupsdisable groucho
$ sudo reject corbet
```

Which to use? It’s a bad idea to accept print jobs that have no hope of being printed in the foreseeable future, so use `reject` for extended downtime. For brief interruptions that should be invisible to users (e.g., changing a toner cartridge), use `cupsdisable`.

Administrators occasionally ask for a mnemonic to help them remember which commands control which end of the queue. Consider: if CUPS “rejects” a job, that means you can’t “inject” it. Another way to keep the commands straight is to remember that accepting and rejecting are things you can do to print jobs, whereas disabling and enabling are things you can do to printers. It doesn’t make any sense to “accept” a printer or queue.

CUPS itself sometimes temporarily disables a printer that it’s having trouble with (e.g., if someone has dislodged a cable). Once you fix the problem, remember to re-`cupsenable` the queue. If you forget, `lpstat` will tell you. (For a complete discussion of this issue and an alternative approach, see [linuxprinting.org/beh.html](https://www.linuxprinting.org/beh.html).)
Other configuration tasks

Today’s printers are eminently configurable, and CUPS lets you tweak a wide variety of features through its web interface and through the `lpadmin` and `lpoptions` commands. As a rule of thumb, `lpadmin` is for system-wide tasks and `lpoptions` is for per-user tasks.

`lpadmin` can restrict access to printers and queues. For example, you can set up printing quotas and specify which users can print to which printers.

Table 12.1 lists the commands that come with CUPS and classifies them according to their origin.

### Table 12.1: CUPS command-line utilities and their origins

<table>
<thead>
<tr>
<th>Command</th>
<th>Function</th>
</tr>
</thead>
<tbody>
<tr>
<td><strong>CUPS</strong></td>
<td></td>
</tr>
<tr>
<td><code>cups-config</code></td>
<td>Prints API, compiler, directory, and link information</td>
</tr>
<tr>
<td><code>cupsdisable</code>&lt;sup&gt;a&lt;/sup&gt;</td>
<td>Stops printing on a printer</td>
</tr>
<tr>
<td><code>cupsenable</code>&lt;sup&gt;a&lt;/sup&gt;</td>
<td>Restarts printing on a printer</td>
</tr>
<tr>
<td><code>lpinfo</code></td>
<td>Shows available devices or drivers</td>
</tr>
<tr>
<td><code>lpoptions</code></td>
<td>Displays or sets printer options and defaults</td>
</tr>
<tr>
<td><code>lpasswd</code></td>
<td>Adds, changes, or deletes digest passwords</td>
</tr>
<tr>
<td><strong>System V</strong></td>
<td></td>
</tr>
<tr>
<td><code>accept, reject</code></td>
<td>Accepts or rejects queue submissions</td>
</tr>
<tr>
<td><code>cancel</code></td>
<td>Cancels print jobs</td>
</tr>
<tr>
<td><code>lp</code></td>
<td>Queues jobs for printing</td>
</tr>
<tr>
<td><code>lpadmin</code></td>
<td>Configures printers</td>
</tr>
<tr>
<td><code>lpmove</code></td>
<td>Moves an existing print job to a new destination</td>
</tr>
<tr>
<td><code>lpstat</code></td>
<td>Prints status information</td>
</tr>
<tr>
<td><strong>BSD</strong></td>
<td></td>
</tr>
<tr>
<td><code>lpc</code></td>
<td>Acts as a general printer-control program</td>
</tr>
<tr>
<td><code>lpq</code></td>
<td>Displays print queues</td>
</tr>
<tr>
<td><code>lpr</code></td>
<td>Queues jobs for printing</td>
</tr>
<tr>
<td><code>lpqm</code></td>
<td>Cancels print jobs</td>
</tr>
</tbody>
</table>

<sup>a</sup> These are actually just the `disable` and `enable` commands from System V, renamed.
12.3 Troubleshooting tips

Printers combine all the foibles of a mechanical device with all the communication eccentricities of a foreign operating system. They (and the software that drives them) seem dedicated to creating problems for you and your users. The next sections offer some general tips for dealing with printer adversity.
Print daemon restart

Always remember to restart daemons after changing a configuration file.

You can restart **cupsd** in whatever way your system normally restarts daemons, usually **systemctl restart org.cups.cupsd.service** or a similar incantation. In theory, you can also send **cupsd** a HUP signal. Alternatively, you can use the CUPS GUI.
Log files

CUPS maintains three logs: a page log, an access log, and an error log. The page log lists the pages that CUPS has printed. The other two logs are just like the access log and error log for Apache, which should not be surprising since the CUPS server is a web server.

The `cupsd.conf` file specifies the logging level and the locations of the log files. They’re all typically kept underneath `/var/log`.

Here’s an excerpt from a log file that corresponds to a single print job:

```
```
Direct printing connections

Under CUPS, to verify the physical connection to a local printer, you can directly run the printer’s back end. For example, here’s what we get when we execute the back end for a USB-connected printer:

```
$ /usr/lib/cups/backend/usb
direct usb "Unknown" "USB Printer (usb)"
direct usb://XEROX/Phaser%206120?serial=YGG210547 "XEROX Phaser 6120" "Phaser 6120"
```

When the USB cable for the Phaser 6120 is disconnected, that printer drops out of the back end’s output:

```
$ /usr/lib/cups/backend/usb
direct usb "Unknown" "USB Printer (usb)"
```
Network printing problems

To begin tracking down a network printing problem, first try connecting to the printer daemon. You can connect to `cupsd` with a web browser (`hostname:631`) or with the `telnet` command (`telnet hostname 631`).

If you have problems debugging a network printer connection, keep in mind that there must be a queue for the job on some machine, a way to decide where to send the job, and a method of sending the job to the machine that hosts the print queue. On the print server, there must be a place to queue the job, sufficient permissions to allow the job to be printed, and a way to output to the device.

Any and all of these prerequisites will, at some point, go awry. Be prepared to hunt for problems in many places, including these:

- System log files on the sending machine, for name resolution and permission problems
- System log files on the print server, for permission problems
- Log files on the sending machine, for missing filters, unknown printers, missing directories, etc.
- The print daemon’s log files on the print server’s machine, for messages about bad device names, incorrect formats, etc.
- The printer log file on the printing machine, for errors in transmitting the job
- The printer log file on the sending machine, for errors about preprocessing or queuing the job

The locations of CUPS log files are specified in `/etc/cups/cupsd.conf`. See Chapter 10, *Logging*, for general information about log management.
### 12.4 Recommended reading

CUPS comes with a lot of documentation in HTML format. An excellent way to access it is to connect to a CUPS server and click the link for on-line help. Of course, this isn’t any help if you’re consulting the documentation to figure out why you can’t connect to the CUPS server. On your computer, the documents should be installed in `/usr/share/doc/cups` in both HTML and PDF formats. If they aren’t there, ask your distribution’s package manager or look on cups.org.

SECTION TWO
NETWORKING
It would be hard to overstate the importance of networks to modern computing, although that
doesn’t seem to stop people from trying. At many sites—perhaps even the majority—web and
email access are the primary uses of computers. As of 2017, internetworldstats.com estimates the
Internet to have more than 3.7 billion users, or just slightly less than half of the world’s
population. In North America, Internet penetration approaches 90%.

TCP/IP (Transmission Control Protocol/Internet Protocol) is the networking system that
underlies the Internet. TCP/IP does not depend on any particular hardware or operating system,
so devices that speak TCP/IP can all exchange data (“interoperate”) despite their many
differences.

TCP/IP works on networks of any size or topology, whether or not they are connected to the
outside world. This chapter introduces the TCP/IP protocols in the context of the Internet, but
stand-alone networks are quite similar at the TCP/IP level.
13.1 TCP/IP AND ITS RELATIONSHIP TO THE INTERNET

TCP/IP and the Internet share a history that goes back multiple decades. The technical success of the Internet is due largely to the elegant and flexible design of TCP/IP and its open and nonproprietary protocol suite. In turn, the leverage provided by the Internet has helped TCP/IP prevail over several competing protocol suites that were favored at one time or another for political or commercial reasons.

The progenitor of the modern Internet was a research network called ARPANET, established in 1969 by the U.S. Department of Defense. By the end of the 1980s the network was no longer a research project and we transitioned to the commercial Internet. Today’s Internet is a collection of private networks owned by Internet service providers (ISPs) that interconnect at many so-called peering points.
Who runs the Internet?

Oversight of the Internet and the Internet protocols has long been a cooperative and open effort, but its exact structure has changed as the Internet has evolved into a public utility and a driving force in the world economy. Current Internet governance is split roughly into administrative, technical, and political wings, but the boundaries between these functions are often vague. The major players are listed below:

- **ICANN**, the Internet Corporation for Assigned Names and Numbers: if any one group can be said to be in charge of the Internet, this is probably it. It’s the only group with any sort of actual enforcement capability. ICANN controls the allocation of Internet addresses and domain names, along with various other snippets such as protocol port numbers. It is organized as a nonprofit corporation headquartered in California. (icann.org)

- **ISOC**, the Internet Society: ISOC is an open-membership organization that represents Internet users. Although it has educational and policy functions, it’s best known as the umbrella organization for the technical development of the Internet. In particular, it is the parent organization of the Internet Engineering Task Force (ietf.org), which oversees most technical work. ISOC is an international nonprofit organization with offices in Washington, D.C. and Geneva. (isoc.org)

- **IGF**, the Internet Governance Forum: a relative newcomer, the IGF was created by the United Nations in 2006 to establish a home for international and policy-oriented discussions related to the Internet. It’s currently structured as a yearly conference series, but its importance is likely to grow over time as governments attempt to exert more control over the operation of the Internet. (intgovforum.org)

Of these groups, ICANN has the toughest job: establishing itself as the authority in charge of the Internet, undoing the mistakes of the past, and foreseeing the future, all while keeping users, governments, and business interests happy.
Network standards and documentation

If your eyes haven’t glazed over just from reading the title of this section, you’ve probably already had several cups of coffee. Nonetheless, accessing the Internet’s authoritative technical documentation is a crucial skill for system administrators, and it’s more entertaining than it sounds.

The technical activities of the Internet community are summarized in documents known as Requests for Comments or RFCs. Protocol standards, proposed changes, and informational bulletins all usually end up as RFCs. RFCs start their lives as Internet Drafts, and after lots of email wrangling and IETF meetings they either die or are promoted to the RFC series. Anyone who has comments on a draft or proposed RFC is encouraged to reply. In addition to standardizing the Internet protocols, the RFC mechanism sometimes just documents or explains aspects of existing practice.

RFCs are numbered sequentially; currently, there are about 8,200. RFCs also have descriptive titles (e.g., *Algorithms for Synchronizing Network Clocks*), but to forestall ambiguity they are usually cited by number. Once distributed, the contents of an RFC are never changed. Updates are distributed as new RFCs with their own reference numbers. Updates may either extend and clarify existing RFCs or supersede them entirely.

RFCs are available from numerous sources, but rfc-editor.org is dispatch central and will always have the most up-to-date information. Look up the status of an RFC at rfc-editor.org before investing the time to read it; it may no longer be the most current document on that subject.

The Internet standards process itself is detailed in RFC2026. Another useful meta-RFC is RFC5540, *40 Years of RFCs*, which describes some of the cultural and technical context of the RFC system.

Don’t be scared away by the wealth of technical detail found in RFCs. Most contain introductions, summaries, and rationales that are useful for system administrators even when the technical details are not. Some RFCs are specifically written as overviews or general introductions. RFCs might not be the gentlest way to learn about a topic, but they are authoritative, concise, and free.

Not all RFCs are full of boring technical details. Here are some of our favorites on the lighter side (usually written on April 1st):

- RFC1149 – *Standard for Transmission of IP Datagrams on Avian Carriers*
- RFC1925 – *The Twelve Networking Truths*
- RFC3251 – *Electricity over IP*
- RFC4041 – *Requirements for Morality Sections in Routing Area Drafts*
• RFC6214 – Adaptation of RFC1149 for IPv6
• RFC6921 – Design Considerations for Faster-Than-Light Communication
• RFC7511 – Scenic Routing for IPv6

In addition to being assigned its own serial number, an RFC can also be assigned an FYI (For Your Information) number, a BCP (Best Current Practice) number, or a STD (Standard) number. FYIs, STDs, and BCPs are subseries of the RFCs that include documents of special interest or importance.

FYIs are introductory or informational documents intended for a broad audience. They can be a good place to start research on an unfamiliar topic if you can find one that’s relevant. Unfortunately, this series has languished recently and not many of the FYIs are up to date.

BCPs document recommended procedures for Internet sites. They consist of administrative suggestions and for system administrators are often the most valuable of the RFC subseries.

STDs document Internet protocols that have completed the IETF’s review and testing process and have been formally adopted as standards.

RFCs, FYIs, BCPs, and STDs are numbered sequentially within their own series, so a document can bear several different identifying numbers. For example, RFC1713, Tools for DNS Debugging, is also known as FYI27.
13.2 **Networking Basics**

Now that we’ve provided a bit of context, let’s look at the TCP/IP protocols themselves. TCP/IP is a protocol “suite,” a set of network protocols designed to work smoothly together. It includes several components, each defined by a standards-track RFC or series of RFCs:

- IP, the Internet Protocol, which routes data packets from one machine to another (RFC791)
- ICMP, the Internet Control Message Protocol, which defines several kinds of low-level support for IP, including error messages, routing assistance, and debugging help (RFC792)
- ARP, the Address Resolution Protocol, which translates IP addresses to hardware addresses (RFC826; ARP can be used with other protocol suites, but it’s an integral part of the way TCP/IP works on most LAN media.)
- UDP, the User Datagram Protocol, which implements unverified, one-way data delivery (RFC768)
- TCP, the Transmission Control Protocol, which implements reliable, full duplex, flow-controlled, error-corrected conversations (RFC793)

These protocols are arranged in a hierarchy or “stack”, with the higher-level protocols making use of the protocols beneath them. TCP/IP is conventionally described as a five-layer system (as shown in **Exhibit A**), but the actual TCP/IP protocols inhabit only three of these layers.

**Exhibit A: TCP/IP layering model**
**IPv4 and IPv6**

The version of TCP/IP that has been in widespread use for nearly five decades is protocol revision 4, aka IPv4. It uses 4-byte IP addresses. A modernized version, IPv6, expands the IP address space to 16 bytes and incorporates several other lessons learned from the use of IPv4. It removes several features of IP that experience has shown to be of little value, making the protocol potentially faster and easier to implement. IPv6 also integrates security and authentication into the basic protocol.

Operating systems and network devices have supported IPv6 for a long time. Google reports statistics about its clients’ use of IPv6 at [google.com/ipv6](http://google.com/ipv6). As of March 2017, the fraction of peers using IPv6 to contact Google sites has risen to about 14% world-wide. In the United States, it’s over 30%.

Those numbers look healthy, but in fact they’re perhaps a bit deceptive because most mobile devices default to IPv6 when they’re on the carrier’s data network, and there are a lot of phones out there. Home and enterprise networks remain overwhelmingly centered on IPv4.

The development and deployment of IPv6 were to a large extent motivated by the concern that the world was running out of 4-byte IPv4 address space. And indeed that concern proved well founded: at this point, only Africa has any remaining IPv4 addresses still available for assignment (see ipv4.potaroo.net for details). The Asia-Pacific region was the first to run out of addresses (on April 19, 2011).

Given that we’ve already lived through the IPv4 apocalypse and have used up all our IPv4 addresses, how is it that the world continues to rely predominantly on IPv4?

For the most part, we’ve learned to make more efficient use of the IPv4 addresses that we have. Network Address Translation (NAT; see [this page](http://this page)) lets entire networks of machines hide behind a single IPv4 address. Classless Inter-Domain Routing (CIDR; see [this page](http://this page)) flexibly subdivides networks and promotes efficient backbone routing. Contention for IPv4 addresses still exists, but like broadcast spectrum, it tends to be reallocated in economic rather than technological ways these days.

The underlying issue that limits IPv6’s adoption is that IPv4 support remains mandatory for a device to be a functional citizen of the Internet. For example, here are a few major web sites that as of 2017 are still not reachable through IPv6: Amazon, Reddit, eBay, IMDB, Hotmail, Tumblr, MSN, Apple, The New York Times, Twitter, Pinterest, Bing, WordPress, Dropbox, craigslist, Stack Overflow. We could go on, but you get the drift.

This list consists of sites whose primary web addresses are not associated with any IPv6 addresses (AAAA records) in DNS. Microsoft Bing’s presence on the list is particularly interesting given that it’s one of a handful of major sites showcased in materials for the World IPv6 Launch marketing campaign of 2012 (tag line: “This time it is for real”). We don’t know
the full story behind this situation, but Bing evidently supported IPv6 at one point, then later decided it wasn’t worth the trouble. See worldipv6launch.org.

Your choice is not between IPv4 and IPv6; it’s between supporting IPv4 alone and supporting both IPv4 and IPv6. When all the services listed above—and scores more in the second tier—have added IPv6 support, then you can reasonably consider adopting IPv6 instead of IPv4. Until then, it doesn’t seem unreasonable to ask IPv6 to justify the effort of its implementation by providing better performance, security, or features. Or perhaps, by opening the door to a world of IPv6-only services that simply can’t be accessed through IPv4.

Unfortunately, those services don’t exist, and IPv6 doesn’t actually offer any of those benefits. Yes, it’s an elegant and well-designed protocol that improves on IPv4. And yes, it is in some ways easier to administer than IPv4 and requires fewer hacks (e.g., less need for NAT). But in the end, it’s just a cleaned-up version of IPv4 with a larger address space. The fact that you must manage it alongside IPv4 eliminates any potential efficiency gain. IPv6’s raison d’être remains the millennial fear of IPv4 address exhaustion, and to date, the effects of that exhaustion just haven’t been painful enough to motivate widespread migration to IPv6.

We’ve been publishing this book for a long time, and over the last few editions, IPv6 has always seemed like it was one more update away from meriting coverage as a primary technology. 2017 brings an uncanny sense of déjà vu, with IPv6 looming ever brighter on the horizon but still solving no immediate problems and offering few specific incentives to convert. IPv6 is the future of networking, and evidently, it always will be.

The arguments in favor of actually deploying IPv6 inside your network remain largely attitudinal: It will have to be done at some point. IPv6 is superior from an engineering standpoint. You need to develop IPv6 expertise so that you’re not caught flat-footed when the IPv6 rapture finally arrives. All the cool kids are doing it.

We say: sure, go ahead, support IPv6 if you feel like it. That’s a responsible and forward-thinking path. It’s civic-minded, too—your adoption of IPv6 hastens the day when IPv6 is all we have to deal with. But if you don’t feel like diving into IPv6, that’s fine too. You’ll have years of warning before there’s any real need to transition.

Of course, none of these comments apply if your organization offers public services on the Internet. In that case, it’s your solemn duty to implement IPv6. Don’t screw things up for the rest of us by continuing to impede IPv6’s adoption. Do you want to be Google, or do you want to be Microsoft Bing?

There’s also an argument to be made for IPv6 in data centers where direct connectivity to the outside world of IPv4 is not needed. In these limited environments, you may indeed have the option to migrate to IPv6 and leave IPv4 behind, thereby simplifying your infrastructure. Internet-facing servers can speak IPv4 even as they route all internal and back-end traffic over IPv6.

A couple of points:
IPv6 has been production-ready for a long time. Implementation bugs aren’t a major concern. Expect it to work as reliably as IPv4.

From a hardware standpoint, IPv6 support should be considered mandatory for all new device acquisitions. It’s doubtful that you could find any piece of enterprise-grade networking gear that doesn’t support IPv6 these days, but a lot of consumer-grade equipment remains IPv4-only.

In this book, we focus on IPv4 as the mainstream version of TCP/IP. IPv6-specific material is explicitly marked. Fortunately for sysadmins, IPv4 and IPv6 are highly analogous. If you understand IPv4, you already know most of what you need to know about IPv6. The main difference between the versions lies in their addressing schemes. In addition to longer addresses, IPv6 introduces a few additional addressing concepts and some new notation. But that’s about it.
**Packets and encapsulation**

TCP/IP supports a variety of physical networks and transport systems, including Ethernet, token ring, MPLS (Multiprotocol Label Switching), wireless Ethernet, and serial-line-based systems. Hardware is managed within the link layer of the TCP/IP architecture, and higher-level protocols do not know or care about the specific hardware being used.

Data travels on a network in the form of packets, bursts of data with a maximum length imposed by the link layer. Each packet consists of a header and a payload. The header tells where the packet came from and where it’s going. It can also include checksums, protocol-specific information, or other handling instructions. The payload is the data to be transferred.

The name of the primitive data unit depends on the layer of the protocol. At the link layer it is called a frame, at the IP layer a packet, and at the TCP layer a segment. In this book, we use “packet” as a generic term that encompasses these various cases.

As a packet travels down the protocol stack (from TCP or UDP transport to IP to Ethernet to the physical wire) in preparation for being sent, each protocol adds its own header information. Each protocol’s finished packet becomes the payload part of the packet generated by the next protocol. This nesting is known as encapsulation. On the receiving machine, the encapsulation is reversed as the packet travels back up the protocol stack.

For example, a UDP packet being transmitted over Ethernet contains three different wrappers or envelopes. On the Ethernet wire, it is framed with a simple header that lists the source and next-hop destination hardware addresses, the length of the frame, and the frame’s checksum (CRC). The Ethernet frame’s payload is an IP packet, the IP packet’s payload is a UDP packet, and the UDP packet’s payload is the data being transmitted. **Exhibit B** shows the components of such a frame.

---

**Exhibit B: A typical network packet**

<table>
<thead>
<tr>
<th>Ethernet header</th>
<th>IPv4 header</th>
<th>UDP header</th>
<th>Application data</th>
<th>Ethernet CRC</th>
</tr>
</thead>
<tbody>
<tr>
<td>14 bytes</td>
<td>20 bytes</td>
<td>8 bytes</td>
<td>100 bytes</td>
<td>4 bytes</td>
</tr>
<tr>
<td><strong>UDP packet (108 bytes)</strong></td>
<td></td>
<td></td>
<td></td>
<td></td>
</tr>
<tr>
<td><strong>IPv4 packet (128 bytes)</strong></td>
<td></td>
<td></td>
<td></td>
<td></td>
</tr>
<tr>
<td><strong>Ethernet frame (146 bytes)</strong></td>
<td></td>
<td></td>
<td></td>
<td></td>
</tr>
</tbody>
</table>
**Ethernet framing**

One of the main chores of the link layer is to add headers to packets and to put separators between them. The headers contain each packet’s link-layer addressing information and checksums, and the separators ensure that receivers can tell where one packet stops and the next one begins. The process of adding these extra bits is known generically as framing.

The link layer is divided into two parts: MAC, the Media Access Control sublayer, and LLC, the Logical Link Control sublayer. The MAC sublayer deals with the media and transmits packets onto the wire. The LLC sublayer handles the framing.

Today, a single standard for Ethernet framing is in common use: DIX Ethernet II. In the distant past, several slightly different standards based on IEEE 802.2 were also used. You might still run across vestigial references to framing choices in network documentation, but you can now ignore this issue.
Maximum transfer unit

The size of packets on a network can be limited both by hardware specifications and by protocol conventions. For example, the payload of a standard Ethernet frame is traditionally 1,500 bytes. The size limit is associated with the link-layer protocol and is called the maximum transfer unit or MTU. Table 13.1 shows some typical values for the MTU.

<table>
<thead>
<tr>
<th>Network type</th>
<th>Maximum transfer unit</th>
</tr>
</thead>
<tbody>
<tr>
<td>Ethernet</td>
<td>1,500 bytes (1,492 with 802.2 framing)</td>
</tr>
<tr>
<td>IPv6 (all hardware)</td>
<td>At least 1,280 bytes at the IP layer</td>
</tr>
<tr>
<td>Token ring</td>
<td>Configurable[a]</td>
</tr>
<tr>
<td>Point-to-point WAN links (T1, T3)</td>
<td>Configurable, often 1,500 or 4,500 bytes</td>
</tr>
</tbody>
</table>

Table 13.1: MTUs for various types of network

IPv4 splits packets to conform to the MTU of a particular network link. If a packet is routed through several networks, one of the intermediate networks might have a smaller MTU than the network of origin. In this case, an IPv4 router that forwards the packet onto the small-MTU network further subdivides the packet in a process called fragmentation.

Fragmentation of in-flight packets is an unwelcome chore for a busy router, so IPv6 largely removes this feature. Packets can still be fragmented, but the originating host must do the work itself. All IPv6 networks are required to support an MTU of at least 1,280 bytes at the IP layer, so IPv6 senders also have the option of limiting themselves to packets of this size.

IPv4 senders can discover the lowest-MTU link through which a packet must pass by setting the packet’s “do not fragment” flag. If the packet reaches an intermediate router that cannot forward the packet without fragmenting it, the router returns an ICMP error message to the sender. The ICMP packet includes the MTU of the network that’s demanding smaller packets, and this MTU then becomes the governing packet size for communication with that destination.

IPv6 path MTU discovery works similarly, but since intermediate routers are never allowed to fragment IPv6 packets, all IPv6 packets act as if they had a “do not fragment” flag enabled. Any IPv6 packet that’s too large to fit into a downstream pipe causes an ICMP message to be returned to the sender.

The TCP protocol automatically does path MTU discovery, even in IPv4. UDP is not so nice and is happy to shunt extra work to the IP layer.

IPv4 fragmentation problems can be insidious. Although path MTU discovery should automatically resolve MTU conflicts, an administrator must occasionally intervene. If you are using a tunneled architecture for a virtual private network, for example, you should look at the
size of the packets that are traversing the tunnel. They are often 1,500 bytes to start with, but once the tunneling header is added, they become 1,540 bytes or so and must be fragmented. Setting the MTU of the link to a smaller value averts fragmentation and increases the overall performance of the tunneled network. Consult the `ifconfig` or `ip-link` man page to see how to set an interface’s MTU.
13.3 **Packet Addressing**

Like letters or email messages, network packets must be properly addressed to reach their destinations. Several addressing schemes are used in combination:

- MAC (Media Access Control) addresses for use by hardware
- IPv4 and IPv6 network addresses for use by software
- Hostnames for use by people
Hardware (MAC) addressing

Each of a host’s network interfaces usually has one link-layer MAC address that distinguishes it from other machines on the physical network, plus one or more IP addresses that identify the interface on the global Internet. This last part bears repeating: IP addresses identify network interfaces, not machines. (To users the distinction is irrelevant, but administrators must know the truth.)

The lowest level of addressing is dictated by network hardware. For example, Ethernet devices are assigned a unique 6-byte hardware address at the time of manufacture. These addresses are traditionally written as a series of 2-digit hex bytes separated by colons; for example, 00:50:8d:9a:3b:df.

Token ring interfaces have a similar address that is also six bytes long. Some point-to-point networks (such as PPP) need no hardware addresses at all; the identity of the destination is specified as the link is established.

A 6-byte Ethernet address is divided into two parts. The first three bytes identify the manufacturer of the hardware, and the last three bytes are a unique serial number that the manufacturer assigns. Sysadmins can sometimes identify the brand of machine that is trashing a network by looking up the 3-byte identifier in a table of vendor IDs. The 3-byte codes are actually IEEE Organizationally Unique Identifiers (OUIs), so you can look up them up directly in the IEEE’s database at

standards.ieee.org/regauth/oui

Of course, the relationships among the manufacturers of chipsets, components, and systems are complex, so the vendor ID embedded in a MAC address can be misleading, too.

In theory, Ethernet hardware addresses are permanently assigned and immutable. However, many network interfaces let you override the hardware address and set one of your own choosing. This feature can be handy if you have to replace a broken machine or network card and for some reason must use the old MAC address (e.g., all your switches filter it, or your DHCP server hands out addresses according to MAC addresses, or your MAC address is also a software license key). Spoofable MAC addresses are also helpful if you need to infiltrate a wireless network that uses MAC-based access control. But for simplicity, it’s generally advisable to preserve the uniqueness of MAC addresses.
IP addressing

See this page for more details about NAT and private address spaces.

At the next level up from the hardware, Internet addressing (more commonly known as IP addressing) is used. IP addresses are hardware independent. Within any particular network context, an IP address identifies a specific and unique destination. However, it’s not quite accurate to say that IP addresses are globally unique because several special cases muddy the water: NAT uses one interface’s IP address to handle traffic for multiple machines; IP private address spaces are addresses that multiple sites can use at once, as long as the addresses are not visible to the Internet; anycast addressing shares one IP address among several machines.

See this page for more information about ARP.

The mapping from IP addresses to hardware addresses is implemented at the link layer of the TCP/IP model. On networks such as Ethernet that support broadcasting (that is, networks that allow packets to be addressed to “all hosts on this physical network”), senders use the ARP protocol to discover mappings without assistance from a system administrator. In IPv6, an interface’s MAC address is often used as part of the IP address, making the translation between IP and hardware addressing virtually automatic.
Hostname “addressing”

IP addresses are sequences of numbers, so they are hard for people to remember. Operating systems allow one or more hostnames to be associated with an IP address so that users can type rfc-editor.org instead of 4.31.198.49. This mapping can be set up in several ways, ranging from a static file (/etc/hosts) to the LDAP database system to DNS, the world-wide Domain Name System. Keep in mind that hostnames are really just a convenient shorthand for IP addresses, and as such, they refer to network interfaces rather than computers.

See Chapter 16 for more information about DNS.
Ports

IP addresses identify a machine’s network interfaces, but they are not specific enough to address individual processes or services, many of which might be actively using the network at once. TCP and UDP extend IP addresses with a concept known as a port, a 16-bit number that supplements an IP address to specify a particular communication channel. Valid ports are in the range 1–65,535.

Standard services such as SMTP, SSH, and HTTP associate themselves with “well known” ports defined in /etc/services. Here are some typical entries from the services file:

```
... 
smtpl 25/udp # Simple Mail Transfer
smtp 25/tcp # Simple Mail Transfer
...
domain 53/udp # Domain Name Server
domain 53/tcp # Domain Name Server
...
http 80/udp www www-http # World Wide Web HTTP
http 80/tcp www www-http # World Wide Web HTTP
...
kerberos 88/udp # Kerberos
kerberos 88/tcp # Kerberos
... 
```

The services file is part of the infrastructure. You should never need to modify it, although you can do so if you want to add a nonstandard service. You can find a full list of assigned ports at iana.org/assignments/port-numbers.

Although both TCP and UDP have ports, and those ports have the same sets of potential values, the port spaces are entirely separate and unrelated. Firewalls must be configured separately for each of these protocols.

To help prevent impersonation of system services, UNIX systems restrict programs from binding to port numbers under 1,024 unless they are run as root or have an appropriate Linux capability. Anyone can communicate with a server running on a low port number; the restriction applies only to the program listening on the port.

Today, the privileged port system is as much a nuisance as it is a bulwark against malfeasance. In many cases, it’s more secure to run standard services on unprivileged ports as nonroot users and to forward network traffic to these high-numbered ports through a load balancer or some other type of network appliance. This practice limits the proliferation of unnecessary root privileges and adds an additional layer of abstraction to your infrastructure.
**Address types**

The IP layer defines several broad types of address, some of which have direct counterparts at the link layer:

- **Unicast** – addresses that refer to a single network interface
- **Multicast** – addresses that simultaneously target a group of hosts
- **Broadcast** – addresses that include all hosts on the local subnet
- **Anycast** – addresses that resolve to any one of a group of hosts

Multicast addressing facilitates applications such as video conferencing for which the same set of packets must be sent to all participants. The Internet Group Management Protocol (IGMP) constructs and manages sets of hosts that are treated as one multicast destination.

Multicast is largely unused on today’s Internet, but it’s slightly more mainstream in IPv6. IPv6 broadcast addresses are really just specialized forms of multicast addressing.

Anycast addresses bring load balancing to the network layer by allowing packets to be delivered to whichever of several destinations is closest in terms of network routing. You might expect that they’d be implemented similarly to multicast addresses, but in fact they are more like unicast addresses.

Most of the implementation details for anycast support are handled at the level of routing rather than through IP. The novelty of anycast addressing is really just the relaxation of the traditional requirement that IP addresses identify unique destinations. Anycast addressing is formally described for IPv6, but the same tricks can be applied to IPv4, too—for example, as is done for root DNS name servers.
13.4 IP ADDRESSES: THE GORY DETAILS

With the exception of multicast addresses, Internet addresses consist of a network portion and a host portion. The network portion identifies a logical network to which the address refers, and the host portion identifies a node on that network. In IPv4, addresses are 4 bytes long and the boundary between network and host portions is set administratively. In IPv6, addresses are 16 bytes long and the network portion and host portion are always 8 bytes each.

IPv4 addresses are written as decimal numbers, one for each byte, separated by periods; for example, 209.85.171.147. The leftmost byte is the most significant and is always part of the network portion.

When 127 is the first byte of an address, it denotes the “loopback network,” a fictitious network that has no real hardware interface and only one host. The loopback address 127.0.0.1 always refers to the current host. Its symbolic name is “localhost”. (This is another small violation of IP address uniqueness since every host thinks 127.0.0.1 is a different computer.)

IPv6 addresses and their text-formatted equivalents are a bit more complicated. They’re discussed in the section IPv6 addressing.

An interface’s IP address and other parameters are set with the ip address (Linux) or ifconfig (FreeBSD) command. Details on configuring a network interface start here.
IPv4 address classes

Historically, IP addresses had an inherent “class” that depended on the first bits of the leftmost byte. The class determined which bytes of the address were in the network portion and which were in the host portion. Today, an explicit mask identifies the network portion, and the boundary can fall between any two adjacent bits, not just between bytes. However, the traditional classes are still used as defaults when no explicit division is specified.

Classes A, B, and C denote regular IP addresses. Classes D and E are multicasting and research addresses. Table 13.2 describes the characteristics of each class. The network portion of an address is denoted by N, and the host portion by H.

Table 13.2: Historical IPv4 address classes

<table>
<thead>
<tr>
<th>Class</th>
<th>1st byte(^a)</th>
<th>Format</th>
<th>Comments</th>
</tr>
</thead>
<tbody>
<tr>
<td>A</td>
<td>1-127</td>
<td>N.H.H.H</td>
<td>Very early networks, or reserved for DoD</td>
</tr>
<tr>
<td>B</td>
<td>128-191</td>
<td>N.N.H.H</td>
<td>Large sites, usually subnetted, were hard to get</td>
</tr>
<tr>
<td>C</td>
<td>192-223</td>
<td>N.N.N.H</td>
<td>Were easy to get, often obtained in sets</td>
</tr>
<tr>
<td>D</td>
<td>224-239</td>
<td>–</td>
<td>Multicast addresses, not permanently assigned</td>
</tr>
<tr>
<td>E</td>
<td>240-255</td>
<td>–</td>
<td>Experimental addresses</td>
</tr>
</tbody>
</table>

* a. The value 0 is special and is not used as the first byte of regular IP addresses. The value 127 is reserved for the loopback address.

It’s unusual for a single physical network to have thousands of computers attached to it, so class A and class B addresses (which allow for 16,777,214 hosts and 65,534 hosts per network, respectively) are really quite wasteful. For example, the 127 class A networks use up half the available 4-byte address space. Who knew that IPv4 address space would become so precious!
IPv4 subnetting

To make better use of these addresses, you can now reassign part of the host portion to the network portion by specifying an explicit 4-byte (32-bit) “subnet mask” or “netmask” in which the 1s correspond to the desired network portion and the 0s correspond to the host portion. The 1s must be leftmost and contiguous. At least eight bits must be allocated to the network part and at least two bits to the host part. Ergo, there are really only 22 possible values for an IPv4 netmask.

For example, the four bytes of a class B address would normally be interpreted as N.N.H.H. The implicit netmask for class B is therefore 255.255.0.0 in decimal notation. With a netmask of 255.255.255.0, however, the address would be interpreted as N.N.N.H. Use of the mask turns a single class B network address into 256 distinct class-C-like networks, each of which can support 254 hosts.

See this page for more information about ip and ifconfig.

Netmasks are assigned with the ip or ifconfig command as each network interface is set up. By default, these commands use the inherent class of an address to figure out which bits are in the network part. When you set an explicit mask, you simply override this behavior.

Netmasks that do not end at a byte boundary can be annoying to decode and are often written as /XX, where XX is the number of bits in the network portion of the address. This is sometimes called CIDR (Classless Inter-Domain Routing; see this page) notation. For example, the network address 128.138.243.0/26 refers to the first of four networks whose first bytes are 128.138.243. The other three networks have 64, 128, and 192 as their fourth bytes. The netmask associated with these networks is 255.255.255.192 or 0xFFFFFFC0; in binary, it’s 26 ones followed by 6 zeros. Exhibit C breaks out these numbers in a bit more detail.

Exhibit C: Netmask base conversion

<table>
<thead>
<tr>
<th>IP address</th>
<th>128</th>
<th>.</th>
<th>138</th>
<th>.</th>
<th>243</th>
<th>.</th>
<th>0</th>
</tr>
</thead>
<tbody>
<tr>
<td>Decimal netmask</td>
<td>255</td>
<td>.</td>
<td>255</td>
<td>.</td>
<td>255</td>
<td>.</td>
<td>192</td>
</tr>
<tr>
<td>Hex netmask</td>
<td>f</td>
<td>f</td>
<td>.</td>
<td>f</td>
<td>f</td>
<td>.</td>
<td>c</td>
</tr>
<tr>
<td>Binary netmask</td>
<td>1111</td>
<td>1111</td>
<td>.</td>
<td>1111</td>
<td>1111</td>
<td>.</td>
<td>1100</td>
</tr>
</tbody>
</table>

A /26 network has 6 bits left (32 – 26 = 6) to number hosts. $2^6$ is 64, so the network has 64 potential host addresses. However, it can only accommodate 62 actual hosts because the all-0 and all-1 host addresses are reserved (they are the network and broadcast addresses, respectively).
In our 128.138.243.0/26 example, the extra two bits of network address obtained by subnetting can take on the values 00, 01, 10, and 11. The 128.138.243.0/24 network has thus been divided into four /26 networks:

- 128.138.243.0/26 (0 in decimal is 00000000 in binary)
- 128.138.243.64/26 (64 in decimal is 01000000 in binary)
- 128.138.243.128/26 (128 in decimal is 10000000 in binary)
- 128.138.243.192/26 (192 in decimal is 11000000 in binary)

The boldfaced bits of the last byte of each address are the bits that belong to the network portion of that byte.
Tricks and tools for subnet arithmetic

It’s confusing to do all this bit twiddling in your head, but some tricks can make it simpler. The number of hosts per network and the value of the last byte in the netmask always add up to 256:

\[
\text{last netmask byte} = 256 - \text{net size}
\]

For example, \(256 - 64 = 192\), which is the final byte of the netmask in the preceding example. Another arithmetic fact is that the last byte of an actual network address (as opposed to a netmask) must be evenly divisible by the number of hosts per network. We see this in action in the 128.138.243.0/26 example, where the last bytes of the networks are 0, 64, 128, and 192—all divisible by 64.

Given an IP address (say, 128.138.243.100), we cannot tell without the associated netmask what the network address and broadcast address will be. Table 13.3 shows the possibilities for /16 (the default for a class B address), /24 (a plausible value), and /26 (a reasonable value for a small network).

<table>
<thead>
<tr>
<th>IP address</th>
<th>Netmask</th>
<th>Network</th>
<th>Broadcast</th>
</tr>
</thead>
<tbody>
<tr>
<td>128.138.243.100/16</td>
<td>255.255.0.0</td>
<td>128.138.0.0</td>
<td>128.138.255.255</td>
</tr>
<tr>
<td>128.138.243.100/24</td>
<td>255.255.255.0</td>
<td>128.138.243.0</td>
<td>128.138.243.255</td>
</tr>
</tbody>
</table>

The network address and broadcast address steal two hosts from each network, so it would seem that the smallest meaningful network would have four possible hosts: two real hosts—usually at either end of a point-to-point link—and the network and broadcast addresses. To have four values for hosts requires two bits in the host portion, so such a network would be a /30 network with netmask 255.255.255.252 or 0xFFFFFFFFFC. However, a /31 network is treated as a special case (see RFC3021) and has no network or broadcast address; both of its two addresses are used for hosts, and its netmask is 255.255.255.254.

A handy web site called the IP Calculator by Krischan Jodies (jodies.de/ipcalc) helps with binary/hex/mask arithmetic. IP Calculator displays everything you might need to know about a network address and its netmask, broadcast address, hosts, etc.

A command-line version of the tool, ipcalc, is also available. It’s in the standard repositories for Debian, Ubuntu, and FreeBSD.

Red Hat and CentOS include a similar but unrelated program that’s also called ipcalc. However, it’s relatively useless because it only understands default IP address classes.

Here’s some sample ipcalc output, munged a bit to help with formatting:
$ ipcalc 24.8.175.69/28
Address: 24.8.175.69 00010000.00001000.10101111.0100 0101
Netmask: 255.255.255.240 = 28 11111111.11111111.11111111.1111 0000
Wildcard: 0.0.0.15 00000000.00000000.00000000.0000 1111
=>
Network: 24.8.175.64/28 00011000.00001000.10101111.0100 0000
HostMin: 24.8.175.65 00011000.00001000.10101111.0100 0001
HostMax: 24.8.175.78 00011000.00001000.10101111.0100 1110
Broadcast: 24.8.175.79 00011000.00001000.10101111.0100 1111
Hosts/Net: 14 Class A

The output includes both easy-to-understand versions of the addresses and “cut and paste” versions. Very useful.

If a dedicated IP calculator isn’t available, the standard utility bc makes a good backup utility since it can do arithmetic in any base. Set the input and output bases with the ibase and obase directives. Set the obase first; otherwise, it’s interpreted relative to the new ibase.
**CIDR: Classless Inter-Domain Routing**

Like subnetting, of which it is a direct extension, CIDR relies on an explicit netmask to define the boundary between the network and host parts of an address. But unlike subnetting, CIDR allows the network portion to be made *smaller* than would be implied by an address’s implicit class. A short CIDR mask can have the effect of aggregating several networks for purposes of routing. Hence, CIDR is sometimes referred to as supernetting.

*CIDR is defined in RFC4632.*

CIDR simplifies routing information and imposes hierarchy on the routing process. Although CIDR was intended as only an interim solution along the road to IPv6, it has proved to be sufficiently powerful to handle the Internet’s growth problems for more than two decades.

For example, suppose that a site has been given a block of eight class C addresses numbered 192.144.0.0 through 192.144.7.0 (in CIDR notation, 192.144.0.0/21). Internally, the site could use them as

- 1 network of length /21 with 2,046 hosts, netmask 255.255.248.0
- 8 networks of length /24 with 254 hosts each, netmask 255.255.255.0
- 16 networks of length /25 with 126 hosts each, netmask 255.255.255.128
- 32 networks of length /26 with 62 hosts each, netmask 255.255.255.192

and so on. But from the perspective of the Internet, it’s not necessary to have 32, 16, or even 8 routing table entries for these addresses. They all refer to the same organization, and all the packets go to the same ISP. A single routing entry for 192.144.0.0/21 suffices. CIDR makes it easy to sub-allocate portions of addresses and thus increases the number of available addresses manyfold.

Inside your network, you can mix and match regions of different subnet lengths as long as all the pieces fit together without overlaps. This is called variable length subnetting. For example, an ISP with the 192.144.0.0/21 allocation could define some /30 networks for point-to-point customers, some /24s for large customers, and some /27s for smaller folks.

All the hosts on a network must be configured with the same netmask. You can’t tell one host that it is a /24 and another host on the same network that it is a /25.
**Address allocation**

Only network numbers are formally assigned; sites must define their own host numbers to form complete IP addresses. You can subdivide the address space that has been assigned to you into subnets however you like.

Administratively, ICANN (the Internet Corporation for Assigned Names and Numbers) has delegated blocks of addresses to five regional Internet registries, and these regional authorities are responsible for doling out subblocks to ISPs within their regions. These ISPs in turn divide up their blocks and hand out pieces to individual clients. Only large ISPs should ever have to deal directly with one of the ICANN-sponsored address registries.

**Table 13.4** lists the regional registration authorities.

**Table 13.4: Regional Internet registries**

<table>
<thead>
<tr>
<th>Name</th>
<th>Site</th>
<th>Region covered</th>
</tr>
</thead>
<tbody>
<tr>
<td>ARIN</td>
<td>arin.net</td>
<td>North America, part of the Caribbean</td>
</tr>
<tr>
<td>APNIC</td>
<td>apnic.net</td>
<td>Asia/Pacific region, including Australia and New Zealand</td>
</tr>
<tr>
<td>AfriNIC</td>
<td>afrinic.net</td>
<td>Africa</td>
</tr>
<tr>
<td>LACNIC</td>
<td>lacnic.net</td>
<td>Central and South America, part of the Caribbean</td>
</tr>
<tr>
<td>RIPE NCC</td>
<td>ripe.net</td>
<td>Europe and surrounding areas</td>
</tr>
</tbody>
</table>

The delegation from ICANN to regional registries and then to national or regional ISPs has allowed for further aggregation in the backbone routing tables. ISP customers who have been allocated address space within the ISP’s block do not need individual routing entries on the backbone. A single entry for the aggregated block that points to the ISP suffices.
Private addresses and network address translation (NAT)

Another factor that has mitigated the effect of the IPv4 address crisis is the use of private IP address spaces, described in RFC1918. These addresses are used by your site internally but are never shown to the Internet (or at least, not intentionally). A border router translates between your private address space and the address space assigned by your ISP.

RFC1918 sets aside 1 class A network, 16 class B networks, and 256 class C networks that will never be globally allocated and can be used internally by any site. Table 13.5 shows the options. (The “CIDR range” column shows each range in the more compact CIDR notation; it does not add additional information.)

Table 13.5: IP addresses reserved for private use

<table>
<thead>
<tr>
<th>IP class</th>
<th>From</th>
<th>To</th>
<th>CIDR range</th>
</tr>
</thead>
<tbody>
<tr>
<td>Class A</td>
<td>10.0.0.0</td>
<td>10.255.255.255</td>
<td>10.0.0.0/8</td>
</tr>
<tr>
<td>Class B</td>
<td>172.16.0.0</td>
<td>172.31.255.255</td>
<td>172.16.0.0/12</td>
</tr>
<tr>
<td>Class C</td>
<td>192.168.0.0</td>
<td>192.168.255.255</td>
<td>192.168.0.0/16</td>
</tr>
</tbody>
</table>

The original idea was that sites would choose an address class from among these options to fit the size of their organizations. But now that CIDR and subnetting are universal, it probably makes the most sense to use the class A address (subnetted, of course) for all new private networks.

To allow hosts that use these private addresses to talk to the Internet, the site’s border router runs a system called NAT (Network Address Translation). NAT intercepts packets addressed with these internal addresses and rewrites their source addresses, using a valid external IP address and perhaps a different source port number. It also maintains a table of the mappings it has made between internal and external address/port pairs so that the translation can be performed in reverse when answering packets arrive from the Internet.

Many garden-variety “NAT” gateways actually perform Port Address Translation, aka PAT: they use a single external IP address and multiplex connections for many internal clients onto the port space of that single address. For example, this is the default configuration for most of the mass-market routers used with cable modems. In practice, NAT and PAT are similar in terms of their implementation, and both systems are commonly referred to as NAT.

A site that uses NAT must still request a small section of address space from its ISP, but most of the addresses thus obtained are used for NAT mappings and are not assigned to individual hosts. If the site later wants to choose another ISP, only the border router and its NAT configuration need be updated, not the configurations of the individual hosts.
Large organizations that use NAT and RFC1918 addresses must institute some form of central coordination so that all hosts, independently of their department or administrative group, have unique IP addresses. The situation can become complicated when one company that uses RFC1918 address space acquires or merges with another company that’s doing the same thing. Parts of the combined organization must often renumber.

It is possible to have a UNIX or Linux box perform the NAT function, but most sites prefer to delegate this task to their routers or network connection devices. See the vendor-specific sections later in this chapter for details. (Of course, many routers now run embedded Linux kernels. Even so, these dedicated systems are still generally more reliable and more secure than general-purpose computers that also forward packets.)

An incorrect NAT configuration can let private-address-space packets escape onto the Internet. The packets might get to their destinations, but answering packets won’t be able to get back. CAIDA, an organization that collects operational data from the Internet backbone, finds that 0.1% to 0.2% of the packets on the backbone have either private addresses or bad checksums. This sounds like a tiny percentage, but it represents thousands of packets every minute on a busy circuit. See caida.org for other interesting statistics and network measurement tools.

One issue raised by NAT is that an arbitrary host on the Internet cannot initiate connections to your site’s internal machines. To get around this limitation, NAT implementations let you preconfigure externally visible “tunnels” that connect to specific internal hosts and ports. Many routers also support the Universal Plug and Play (UPnP) standards promoted by Microsoft, one feature of which allows interior hosts to set up their own dynamic NAT tunnels. This can be either a godsend or a security risk, depending on your perspective. You can easily disable the feature at the router if you so desire.

Another NAT-related issue is that some applications embed IP addresses in the data portion of packets; these applications are foiled or confused by NAT. Examples include some media streaming systems, routing protocols, and FTP commands. NAT sometimes breaks VPNs, too.

NAT hides interior structure. This secrecy feels like a security win, but the security folks say NAT doesn’t really help for security and does not replace the need for a firewall. Unfortunately, NAT also foils attempts to measure the size and topology of the Internet. See RFC4864, *Local Network Protection for IPv6*, for a good discussion of both the real and illusory benefits of NAT in IPv4.
IPv6 addressing

IPv6 addresses are 128 bits long. These long addresses were originally intended to solve the problem of IP address exhaustion. But now that they’re here, they are being exploited to help with issues of routing, mobility, and locality of reference.

The boundary between the network portion and the host portion of an IPv6 address is fixed at /64, so there can be no disagreement or confusion about how long an address’s network portion “really” is. Stated another way, true subnetting no longer exists in the IPv6 world, although the term “subnet” lives on as a synonym for “local network.” Even though network numbers are always 64 bits long, routers needn’t pay attention to all 64 bits when making routing decisions. They can route packets by prefix, just as they do under CIDR.

IPv6 address notation

The standard notation for IPv6 addresses divides the 128 bits of an address into 8 groups of 16 bits each, separated by colons. For example,

2607:f8b0:000a:0806:0000:0000:0000:200e

This is a real IPv6 address, so don’t use it on your own systems, even for experimentation. RFC3849 suggests that documentation and examples show IPv6 addresses within the prefix block 2001:db8::/32. But we wanted to show a real example that’s routed on the Internet backbone.

Each 16-bit group is represented by 4 hexadecimal digits. This is different from IPv4 notation, in which each byte of the address is represented by a decimal (base 10) number.

A couple of notational simplifications help limit the amount of typing needed to represent IPv6 addresses. First, you needn’t include leading zeros within a group. 000a in the third group above can be written simply as a, and 0806 in the fourth group can be written as 806. Groups with a value of 0000 should be represented as 0. Application of this rule reduces the address above to the following string:

2607:f8b0:a:806:0:0:0:200e

Second, you can replace any number of contiguous, zero-valued, 16-bit groups with a double colon:

2607:f8b0:a:806::200e

The :: can be used only once within an address. However, it can appear as the first or last component. For example, the IPv6 loopback address (analogous to 127.0.0.1 in IPv4) is ::1, which is equivalent to 0:0:0:0:0:0:0:1.

The original specification for IPv6 addresses, RFC4921, documented these notational simplifications but did not require their use. As a result, there can be multiple RFC491-compliant
ways to write a given IPv6 address, as illustrated by the several versions of the example address above.

This polymorphousness makes searching and matching difficult because addresses must be normalized before they can be compared. That’s a problem: we can’t expect standard data-wrangling software such as spreadsheets, scripting languages, and databases to know about the details of IPv6 notation.

RFC5952 updates RFC4921 to make the notational simplifications mandatory. It also adds a few more rules to ensure that every address has only a single text representation:

- Hex digits a–f must be represented by lowercase letters.
- The :: element cannot replace a single 16-bit group. (Just use :0:.)
- If there is a choice of groups to replace with ::, the :: must replace the longest possible sequence of zeros.

You will still see RFC5952-noncompliant addresses out in the wild, and nearly all networking software accepts them, too. However, we strongly recommend following the RFC5952 rules in your configurations, recordkeeping, and software.

IPv6 prefixes

IPv4 addresses were not designed to be geographically clustered in the manner of phone numbers or zip codes, but clustering was added after the fact in the form of the CIDR conventions. (Of course, the relevant “geography” is really routing space rather than physical location.) CIDR was so technically successful that hierarchical subassignment of network addresses is now assumed throughout IPv6.

Your IPv6 ISP obtains blocks of IPv6 prefixes from one of the regional registries listed in Table 13.4. The ISP in turn assigns you a prefix that you prepend to the local parts of your addresses, usually at your border router. Organizations are free to set delegation boundaries wherever they wish within the address spaces assigned to them.

Whenever an address prefix is represented in text form, IPv6 adopts CIDR notation to represent the length of the prefix. The general pattern is

**IPv6-address/prefix-length-in-decimal**

The IPv6-address portion is as outlined in the previous section. It must be a full-length 128-bit address. In most cases, the address bits beyond the prefix are set to zero. However, it’s sometimes appropriate to specify a complete host address along with a prefix length; the intent and meaning are usually clear from context.

The IPv6 address shown in the previous section leads to a Google server. The 32-bit prefix that’s routed on the North American Internet backbone is
In this case, the address prefix was assigned by ARIN directly to Google, as you can verify by looking up the prefix at arin.net. There is no intervening ISP. Google is responsible for structuring the remaining 32 bits of the network number as it sees fit. Most likely, several additional layers of prefixing are used within the Google infrastructure.

In this case, the prefix lengths of the ARIN-assigned address block and the backbone routing table entry are the same, but that is not always true. The allocation prefix determines an administrative boundary, whereas the routing prefix relates to route-space locality.

**Automatic host numbering**

A machine’s 64-bit interface identifier (the host portion of the IPv6 address) can be automatically derived from the interface’s 48-bit MAC (hardware) address with the algorithm known as “modified EUI-64,” documented in RFC4291.

Specifically, the interface identifier is just the MAC address with the two bytes 0xFFFE inserted in the middle and one bit complemented. The bit you will flip is the 7th most significant bit of the first byte; in other words, you XOR the first byte with 0x02. For example, on an interface with MAC address 00:1b:21:30:e9:c7, the autogenerated interface identifier would be 021b:21ff:fe30:e9c7. The underlined digit is 2 instead of 0 because of the flipped bit.

This scheme allows for automatic host numbering, which is a nice feature for sysadmins since only the network portion of addresses need be managed.

That the MAC address can be seen at the IP layer has both good and bad implications. The good part is that host number configuration can be completely automatic. The bad part is that the manufacturer of the interface card is encoded in the first half of the MAC address (see [this page](#)), so you inevitably give up some privacy. Prying eyes and hackers with code for a particular architecture will be helped along. The IPv6 standards point out that sites are not required to use MAC addresses to derive host IDs; they can use whatever numbering system they want.

Virtual servers have virtual network interfaces. The MAC addresses associated with these interfaces are typically randomized, which all but guarantees uniqueness within a particular local context.

**Stateless address autoconfiguration**

The autogenerated host numbers described in the previous section combine with a couple of other simple IPv6 features to enable automatic network configuration for IPv6 interfaces. The overall scheme is known as SLAAC, for StateLess Address AutoConfiguration.

SLAAC configuring for an interface begins by assigning an address on the “link-local network,” which has the fixed network address fe80::/64. The host portion of the address is set from the MAC address of the interface, as described above.
IPv6 does not have IPv4-style broadcast addresses per se, but the link-local network serves roughly the same purpose: it means “this physical network.” Routers never forward packets that were sent to addresses on this network.

Once the link-local address for an interface has been set, the IPv6 protocol stack sends an ICMP Router Solicitation packet to the “all routers” multicast address. Routers respond with ICMP Router Advertisement packets that list the IPv6 network numbers (prefixes, really) in use on the network.

If one of these networks has its “autoconfiguration OK” flag set, the inquiring host assigns an additional address to its interface that combines the network portion advertised by the router with the autogenerated host portion constructed with the modified EUI-64 algorithm. Other fields in the Router Advertisement allow a router to identify itself as an appropriate default gateway and to communicate the network’s MTU.

See this page for more information about DHCP.

The end result is that a new host becomes a full citizen of the IPv6 network without the need for any server (other than the router) to be running on the network and without any local configuration. Unfortunately, the system does not address the configuration of higher-level software such as DNS, so you may still want to run a traditional DHCPv6 server, too.

You will sometimes see IPv6 network autoconfiguration associated with the name Neighbor Discovery Protocol. Although RFC4861 is devoted to the Neighbor Discovery Protocol, the term is actually rather vague. It covers the use and interpretation of a variety of ICMPv6 packet types, some of which are only peripherally related to discovering network neighbors. From a technical perspective, the relationship is that the SLAAC procedure described above uses some, but not all, of the ICMPv6 packet types defined in RFC4861. It’s clearer to just call it SLAAC or “IPv6 autoconfiguration” and to reserve “neighbor discovery” for the IP-to-MAC mapping process described starting here.

**IPv6 tunneling**

Various schemes have been proposed to ease the transition from IPv4 to IPv6, mostly focusing on ways to tunnel IPv6 traffic through the IPv4 network to compensate for gaps in IPv6 support. The two tunneling systems in common use are called 6to4 and Teredo; the latter, named after a family of wood-boring shipworms, can be used on systems behind a NAT device.

**IPv6 information sources**

Here are some useful sources of additional IPv6 information:

- worldipv6launch.com – A variety of IPv6 propaganda
- RFC3587 – IPv6 Global Unicast Address Format
• RFC4291 – *IP Version 6 Addressing Architecture*
13.5 Routing

Routing is the process of directing a packet through the maze of networks that stand between its source and its destination. In the TCP/IP system, it is similar to asking for directions in an unfamiliar country. The first person you talk to might point you toward the right city. Once you were a bit closer to your destination, the next person might be able to tell you how to get to the right street. Eventually, you get close enough that someone can identify the building you’re looking for.

Routing information takes the form of rules (“routes”), such as “To reach network A, send packets through machine C.” There can also be a default route that tells what to do with packets bound for a network to which no explicit route exists.

Routing information is stored in a table in the kernel. Each table entry has several parameters, including a mask for each listed network. To route a packet to a particular address, the kernel picks the most specific of the matching routes—that is, the one with the longest mask. If the kernel finds no relevant route and no default route, then it returns a “network unreachable” ICMP error to the sender.

The word “routing” is commonly used to mean two distinct things:

- Looking up a network address in the routing table as part of the process of forwarding a packet toward its destination
- Building the routing table in the first place

In this section we examine the forwarding function and look at how routes can be manually added to or deleted from the routing table. We defer the more complicated topic of routing protocols that build and maintain the routing table until Chapter 15.
Routing tables

You can examine a machine’s routing table with `ip route show` on Linux or `netstat -r` on FreeBSD. Although `netstat` on Linux is on its way out, it still exists and continues to work. We use `netstat` for the examples below just to avoid having to show two different versions of the output. The `ip` version contains similar content, but its format is somewhat different.

Use `netstat -rn` to avoid DNS lookups and present all information numerically, which is generally more useful. Here is a short example of an IPv4 routing table to give you a better idea of what routes look like:

```
redhat$ netstat -rn

<table>
<thead>
<tr>
<th>Destination</th>
<th>Genmask</th>
<th>Gateway</th>
<th>F1</th>
<th>MSS</th>
<th>Iface</th>
</tr>
</thead>
<tbody>
<tr>
<td>132.236.227.0</td>
<td>255.255.255.0</td>
<td>132.236.227.93</td>
<td>U</td>
<td>1500</td>
<td>eth0</td>
</tr>
<tr>
<td>default</td>
<td>0.0.0.0</td>
<td>132.236.227.1</td>
<td>UG</td>
<td>1500</td>
<td>eth0</td>
</tr>
<tr>
<td>132.236.212.0</td>
<td>255.255.255.192</td>
<td>132.236.212.1</td>
<td>U</td>
<td>1500</td>
<td>eth1</td>
</tr>
<tr>
<td>132.236.220.64</td>
<td>255.255.255.192</td>
<td>132.236.212.6</td>
<td>UG</td>
<td>1500</td>
<td>eth1</td>
</tr>
<tr>
<td>127.0.0.1</td>
<td>255.255.255.255</td>
<td>127.0.0.1</td>
<td>U</td>
<td>3584</td>
<td>lo</td>
</tr>
</tbody>
</table>
```

This host has two network interfaces: 132.236.227.93 (eth0) on the network 132.236.227.0/24 and 132.236.212.1 (eth1) on the network 132.236.212.0/26.

The destination field is usually a network address, although you can also add host-specific routes (their genmask is 255.255.255.255 since all bits are consulted). An entry’s gateway field must contain the full IP address of a local network interface or adjacent host; on Linux kernels it can be 0.0.0.0 to invoke the default gateway.

For example, the fourth route in the table above says that to reach the network 132.236.220.64/26, packets must be sent to the gateway 132.236.212.6 through interface eth1. The second entry is a default route; packets not explicitly addressed to any of the three networks listed (or to the machine itself) are sent to the default gateway host, 132.236.227.1.

A host can route packets only to gateway machines that are reachable through a directly connected network. The local host’s job is limited to moving packets one hop closer to their destinations, so it is pointless to include information about nonadjacent gateways in the local routing table. Each gateway that a packet visits makes a fresh next-hop routing decision by consulting its own local routing database. (The IPv4 source routing feature is an exception to this rule; see this page.)

Routing tables can be configured statically, dynamically, or with a combination of the two approaches. A static route is one that you enter explicitly with the `ip` (Linux) or `route` (FreeBSD) command. Static routes remain in the routing table as long as the system is up; they are often set up at boot time from one of the system startup scripts. For example, the Linux commands

```
ip route add 132.236.220.64/26 via 132.236.212.6 dev eth1
ip route add default via 132.236.227.1 dev eth0
```
add the fourth and second routes displayed by `netstat -rn` above. (The first and third routes in that display were added automatically when the eth0 and eth1 interfaces were configured.) The equivalent FreeBSD commands are similar:

```
route add -net 132.236.220.64/26 gw 132.236.212.6 eth1
route add default gw 132.236.227.1 eth0
```

The final route is also added at boot time. It configures the loopback interface, which prevents packets sent from the host to itself from going out on the network. Instead, they are transferred directly from the network output queue to the network input queue inside the kernel.

In a stable local network, static routing is an efficient solution. It is easy to manage and reliable. However, it requires that the system administrator know the topology of the network accurately at boot time and that the topology not change often.

Most machines on a local area network have only one way to get out to the rest of the network, so the routing problem is easy. A default route added at boot time suffices to point toward the way out. Hosts that use DHCP (see this page) to get their IP addresses can also obtain a default route with DHCP.

For more complicated network topologies, dynamic routing is required. Dynamic routing is implemented by a daemon process that maintains and modifies the routing table. Routing daemons on different hosts communicate to discover the topology of the network and to figure out how to reach distant destinations. Several routing daemons are available. See Chapter 15, *IP Routing*, for details.
ICMP redirects

Although IP generally does not concern itself with the management of routing information, it does define a naïve damage control feature called an ICMP redirect. When a router forwards a packet to a machine on the same network from which the packet was originally received, something is clearly wrong. Since the sender, the router, and the next-hop router are all on the same network, the packet could have been forwarded in one hop rather than two. The router can conclude that the sender’s routing tables are inaccurate or incomplete.

In this situation, the router can notify the sender of its problem by sending an ICMP redirect packet. In effect, a redirect says, “You should not be sending packets for host xxx to me; you should send them to host yyy instead.”

In theory, the recipient of a redirect can adjust its routing table to fix the problem. In practice, redirects contain no authentication information and are therefore untrustworthy. Dedicated routers usually ignore redirects, but most UNIX and Linux systems accept them and act on them by default. You’ll need to consider the possible sources of redirects in your network and disable their acceptance if they could pose a problem.

Under Linux, the variable accept_redirects in the /proc hierarchy controls the acceptance of ICMP redirects. See this page for instructions on examining and resetting this variable.

On FreeBSD the parameters net.inet.icmp.drop_redirect and net.inet6.icmp6.rediraccept control the acceptance of ICMP redirects. Set them to 1 and 0, respectively, in the file /etc/sysctl.conf to ignore redirects. (To activate the new settings, reboot or run sudo /etc/rc.d/sysctl reload.)
13.6 IPv4 ARP and IPv6 Neighbor Discovery

Although IP addresses are hardware-independent, hardware addresses must still be used to actually transport data across a network’s link layer. (An exception is for point-to-point links, where the identity of the destination is sometimes implicit.) IPv4 and IPv6 use separate but eerily similar protocols to discover the hardware address associated with a particular IP address.

IPv4 uses ARP, the Address Resolution Protocol, defined in RFC826. IPv6 uses parts of the Neighbor Discovery Protocol defined in RFC4861. These protocols can be used on any kind of network that supports broadcasting or all-nodes multicasting, but they are most commonly described in terms of Ethernet.

If host A wants to send a packet to host B on the same Ethernet, it uses ARP or ND to discover B’s hardware address. If B is not on the same network as A, host A uses the routing system to determine the next-hop router along the route to B and then uses ARP or ND to find that router’s hardware address. These protocols can only be used to find the hardware addresses of machines that are directly connected to the sending host’s local networks.

Every machine maintains a table in memory called the ARP or ND cache which contains the results of recent queries. Under normal circumstances, many of the addresses a host needs are discovered soon after booting, so ARP and ND do not account for a lot of network traffic.

These protocols work by broadcasting or multicasting a packet of the form “Does anyone know the hardware address for IP address X?” The machine being searched for recognizes its own IP address and replies, “Yes, that’s the IP address assigned to one of my network interfaces, and the corresponding MAC address is 08:00:20:00:fb:6a.”

The original query includes the IP and MAC addresses of the requester so that the machine being sought can reply without issuing a query of its own. Thus, the two machines learn each other’s address mappings with only one exchange of packets. Other machines that overhear the requester’s initial broadcast can record its address mapping, too.

On Linux, the `ip neigh` command examines and manipulates the caches created by ARP and ND, adds or deletes entries, and flushes or prints the table. `ip neigh show` displays the contents of the caches.

On FreeBSD, the `arp` command manipulates the ARP cache and the `ndp` command gives access to the ND cache.

These commands are generally useful only for debugging and for situations that involve special hardware. For example, if two hosts on a network are using the same IP address, one has the right ARP or ND table entry and one is wrong. You can use the cache information to track down the offending machine.
Inaccurate cache entries can be a sign that someone with access to your local network is attempting to hijack network traffic. This type of attack is known generically as ARP spoofing or ARP cache poisoning.
13.7 DHCP: the Dynamic Host Configuration Protocol

DHCP is defined in RFCs 2131, 2132, and 3315.

When you plug a device or computer into a network, it usually obtains an IP address for itself on the local network, sets up an appropriate default route, and connects itself to a local DNS server. The Dynamic Host Configuration Protocol (DHCP) is the hidden Svengali that makes this magic happen.

The protocol lets a DHCP client “lease” a variety of network and administrative parameters from a central server that is authorized to distribute them. The leasing paradigm is particularly convenient for PCs that are turned off when not in use and for networks that must support transient guests such as laptops.

Leasable parameters include

- IP addresses and netmasks
- Gateways (default routes)
- DNS name servers
- Syslog hosts
- WINS servers, X font servers, proxy servers, NTP servers
- TFTP servers (for loading a boot image)

There are dozens more—see RFC2132 for IPv4 and RFC3315 for IPv6. Real-world use of the more exotic parameters is rare, however.

Clients must report back to the DHCP server periodically to renew their leases. If a lease is not renewed, it eventually expires. The DHCP server is then free to assign the address (or whatever was being leased) to a different client. The lease period is configurable, but it’s usually quite long (hours or days).

Even if you want each host to have its own permanent IP address, DHCP can save you time and suffering because it concentrates configuration information on the DHCP server rather than requiring it to be distributed to individual hosts. Once the server is up and running, clients can use DHCP to obtain their network configurations at boot time. The clients needn’t know that they’re receiving a static configuration.
**DHCP software**

ISC, the Internet Systems Consortium, maintains a nice open source reference implementation of DHCP. Major versions 2, 3, and 4 of ISC’s software are all in common use, and all these versions work fine for basic service. Version 3 supports backup DHCP servers, and version 4 supports IPv6. Server, client, and relay agents are all available from isc.org.

Vendors all package some version of the ISC software, although you may have to install the server portion explicitly. The server package is called `dhcp` on Red Hat and CentOS, `isc-dhcp-server` on Debian and Ubuntu, and `isc-dhcp43-server` on FreeBSD. Make sure you’re installing the software you intend, as many systems package multiple implementations of both the server and client sides of DHCP.

It’s best not to tamper with the client side of DHCP, since that part of the code is relatively simple and comes preconfigured and ready to use. Changing the client side of DHCP is not trivial.

However, if you need to run a DHCP server, we recommend the ISC package over vendor-specific implementations. In a typical heterogeneous network environment, administration is greatly simplified by standardizing on a single implementation. The ISC software is a reliable, open source solution that builds without incident on most systems.

Another option to consider is Dnsmasq, a server that implements DHCP service in combination with a DNS forwarder. It’s a tidy package that runs on pretty much any system. The project home page is [thekelleys.org.uk/dnsmasq/doc.html](http://thekelleys.org.uk/dnsmasq/doc.html).

DHCP server software is also built into most routers. Configuration is usually more painful than on a UNIX- or Linux-based server, but reliability and availability might be higher.

In the next few sections, we briefly discuss the DHCP protocol, explain how to set up the ISC server that implements it, and review some client configuration issues.
**DHCP behavior**

DHCP is a backward-compatible extension of BOOTP, a protocol originally devised to help diskless workstations boot. DHCP generalizes the parameters that can be supplied and adds the concept of a lease period for assigned values.

A DHCP client begins its interaction with a DHCP server by broadcasting a “Help! Who am I?” message. IPv4 clients initiate conversations with the DHCP server by using the generic all-1s broadcast address. The clients don’t yet know their subnet masks and therefore can’t use the subnet broadcast address. IPv6 uses multicast addressing instead of broadcasting.

If a DHCP server is present on the local network, it negotiates with the client to supply an IP address and other networking parameters. If there is no DHCP server on the local net, servers on different subnets can receive the initial broadcast message through a separate piece of DHCP software that acts as a relay agent.

When the client’s lease time is half over, it attempts to renew its lease. The server is obliged to keep track of the addresses it has handed out, and this information must persist across reboots. Clients are supposed to keep their lease state across reboots too, although many do not. The goal is to maximize stability in network configuration. In theory, all software should be prepared for network configurations to change at a moment’s notice, but some software still makes unwarranted assumptions about the continuity of the network.
ISC’s DHCP software

ISC’s server daemon is called dhcpd, and its configuration file is dhcpd.conf, usually found in /etc or /etc/dhcp3. The format of the config file is a bit fragile; leave out a semicolon and you may receive a cryptic, unhelpful error message.

When setting up a new DHCP server, you must also make sure that an empty lease database file has been created. Check the summary at the end of the man page for dhcpd to find the correct location for the lease file on your system. It’s usually somewhere underneath /var.

To set up the dhcpd.conf file, you need the following information:

- The subnets for which dhcpd should manage IP addresses, and the ranges of addresses to dole out
- A list of static IP address assignments you want to make (if any), along with the MAC (hardware) addresses of the recipients
- The initial and maximum lease durations, in seconds
- Any other options the server should pass to DHCP clients: netmask, default route, DNS domain, name servers, etc.

The dhcpd man page outlines the configuration process, and the dhcpd.conf man page covers the exact syntax of the config file. In addition to setting up your configuration, make sure dhcpd is started automatically at boot time. (See Chapter 2, Booting and System Management Daemons, for instructions.) It’s helpful to make startup of the daemon conditional on the existence of the dhcpd.conf file if your system doesn’t automatically do this for you.

Below is a sample dhcpd.conf file from a Linux box with two interfaces, one internal and one that connects to the Internet. This machine performs NAT translation for the internal network (see this page) and leases out a range of 10 IP addresses on this network as well.

Every subnet must be declared, even if no DHCP service is provided on it, so this dhcpd.conf file contains a dummy entry for the external interface. It also includes a host entry for one particular machine that needs a fixed address.
### global options

```plaintext
option domain-name "synack.net";
option domain-name-servers gw.synack.net;
option subnet-mask 255.255.255.0;
default-lease-time 600;
max-lease-time 7200;
```

```plaintext
subnet 192.168.1.0 netmask 255.255.255.0 {
    range 192.168.1.51 192.168.1.60;
    option broadcast-address 192.168.1.255;
    option routers gw.synack.net;
}
```

```plaintext
subnet 209.180.251.0 netmask 255.255.255.0 {
}
```

```plaintext
host gandalf {
    hardware ethernet 08:00:07:12:34:56;
    fixed-address gandalf.synack.net;
}
```

---

See [Chapter 16](#) for more information about DNS.

---

Unless you make static IP address assignments such as the one for gandalf above, you need to consider how your DHCP configuration interacts with DNS. The easy option is to assign a generic name to each dynamically leased address (e.g., dhcp1.synack.net) and allow the names of individual machines to float along with their IP addresses. Alternatively, you can configure `dhcppd` to update the DNS database as it hands out addresses. The dynamic update solution is more complicated, but it has the advantage of preserving each machine’s hostname.

ISC’s DHCP relay agent is a separate daemon called `dhcrelay`. It’s a simple program with no configuration file of its own, although vendors often add a startup harness that feeds it the appropriate command-line arguments for your site. `dhcrelay` listens for DHCP requests on local networks and forwards them to a set of remote DHCP servers that you specify. It’s handy both for centralizing the management of DHCP service and for provisioning backup DHCP servers.

ISC’s DHCP client is similarly configuration free. It stores status files for each connection in the directory `/var/lib/dhcp` or `/var/lib/dhclient`. The files are named after the interfaces they describe. For example, `dhclient-eth0.leases` would contain all the networking parameters that `dhclient` had set up on behalf of the eth0 interface.
13.8 Security issues

We address the topic of security in a chapter of its own (Chapter 27), but several security issues relevant to IP networking merit discussion here. In this section, we briefly look at a few networking features that have acquired a reputation for causing security problems, and we recommend ways to minimize their impact. The details of our example systems’ default behavior on these issues (and the appropriate methods for changing them) vary considerably and are discussed in the system-specific material starting here.
**IP forwarding**

A UNIX or Linux system that has IP forwarding enabled can act as a router. That is, it can accept third party packets on one network interface, match them to a gateway or destination host on another interface, and retransmit the packets.

Unless your system has multiple network interfaces and is actually supposed to function as a router, it’s best to turn this feature off. Hosts that forward packets can sometimes be coerced into compromising security by making external packets appear to have come from inside your network. This subterfuge can help an intruder’s packets evade network scanners and packet filters.

It is perfectly acceptable for a host to have network interfaces on multiple subnets and to use them for its own traffic without forwarding third party packets.
ICMP redirects

ICMP redirects (see this page) can maliciously reroute traffic and tamper with your routing tables. Most operating systems listen to ICMP redirects and follow their instructions by default. It would be bad if all your traffic were rerouted to a competitor’s network for a few hours, especially while backups were running! We recommend that you configure your routers (and hosts acting as routers) to ignore and perhaps log ICMP redirect attempts.
Source routing

IPv4’s source routing mechanism lets you specify an explicit series of gateways for a packet to transit on the way to its destination. Source routing bypasses the next-hop routing algorithm that’s normally run at each gateway to determine how a packet should be forwarded.

Source routing was part of the original IP specification; it was intended primarily to facilitate testing. It can create security problems because packets are often filtered according to their origin. If someone can cleverly route a packet to make it appear to have originated within your network instead of the Internet, it might slip through your firewall. We recommend that you neither accept nor forward source-routed packets.

Despite the Internet’s dim view of IPv4 source routing, it somehow managed to sneak its way into IPv6 as well. However, this IPv6 feature was deprecated by RFC5095 in 2007. Compliant IPv6 implementations are now required to reject source-routed packets and return an error message to the sender. Linux and FreeBSD both follow the RFC5095 behavior, as do commercial routers.

Even so, IPv6 source routing may be poised to stage a mini-comeback in the form of “segment routing,” a feature that has now been integrated into the Linux kernel. See lwn.net/Articles/722804 for a discussion of this technology.
Broadcast pings and other directed broadcasts

Ping packets addressed to a network’s broadcast address (instead of to a particular host address) are typically delivered to every host on the network. Such packets have been used in denial-of-service attacks; for example, the so-called Smurf attacks. (The “Smurf attacks” Wikipedia article has details.)

Broadcast pings are a form of “directed broadcast” in that they are packets sent to the broadcast address of a distant network. The default handling of such packets has been gradually changing. For example, versions of Cisco’s IOS up through 11.x forwarded directed broadcast packets by default, but IOS releases since 12.0 do not. It is usually possible to convince your TCP/IP stack to ignore broadcast packets that come from afar, but since this behavior must be set on each interface, the task can be nontrivial at a large site.
**IP spoofing**

The source address on an IP packet is normally filled in by the kernel’s TCP/IP implementation and is the IP address of the host from which the packet was sent. However, if the software creating the packet uses a raw socket, it can fill in any source address it likes. This is called IP spoofing and is usually associated with some kind of malicious network behavior. The machine identified by the spoofed source IP address (if it is a real address at all) is often the victim in the scheme. Error and return packets can disrupt or flood the victim’s network connections. Packet spoofing from a large set of external machines is called a “distributed denial-of-service attack.”

Deny IP spoofing at your border router by blocking outgoing packets whose source address is not within your address space. This precaution is especially important if your site is a university where students like to experiment and might be tempted to carry out digital vendettas.

If you are using private address space internally, you can filter at the same time to catch any internal addresses escaping to the Internet. Such packets can never be answered (because they lack a backbone route) and always indicate that your site has some kind of internal configuration error.

In addition to detecting outbound packets with bogus source addresses, you must also protect against an attacker’s forging the source address on external packets to fool your firewall into thinking that they originated on your internal network. A heuristic known as “unicast reverse path forwarding” (uRPF) helps to address this problem. It makes IP gateways discard packets that arrive on an interface different from the one on which they would be transmitted if the source address were the destination. It’s a quick sanity check that uses the normal IP routing table as a way to validate the origin of network packets. Dedicated routers implement uRPF, but so does the Linux kernel. On Linux, it’s enabled by default.

If your site has multiple connections to the Internet, it might be perfectly reasonable for inbound and outbound routes to be different. In this situation, you’ll have to turn off uRPF to make your routing work properly. If your site has only one way out to the Internet, then turning on uRPF is usually safe and appropriate.
Host-based firewalls

Traditionally, a network packet filter or firewall connects your local network to the outside world and controls traffic according to a site-wide policy. Unfortunately, Microsoft has warped everyone’s perception of how a firewall should work with its notoriously insecure Windows systems. The last few Windows releases all come with their own personal firewalls, and they complain bitterly if you try to turn off the firewall.

Our example systems all include packet filtering software, but you should not infer from this that every UNIX or Linux machine needs its own firewall. The packet filtering features are there primarily to allow these machines to serve as network gateways.

However, we don’t recommend using a workstation as a firewall. Even with meticulous hardening, full-fledged operating systems are too complex to be fully trustworthy. Dedicated network equipment is more predictable and more reliable—even if it secretly runs Linux.

Even sophisticated software solutions like those offered by Check Point (whose products run on UNIX, Linux, and Windows hosts) are not as secure as a dedicated device such as Cisco’s Adaptive Security Appliance series. The software-only solutions are nearly the same price, to boot.

A more thorough discussion of firewall-related issues begins [here](#).
Virtual private networks

Many organizations that have offices in several locations would like to have all those locations connected to one big private network. Such organizations can use the Internet as if it were a private network by establishing a series of secure, encrypted “tunnels” among their various locations. A network that includes such tunnels is known as a virtual private network or VPN.

VPN facilities are also needed when employees must connect to your private network from their homes or from the field. A VPN system doesn’t eliminate every possible security issue relating to such ad hoc connections, but it’s secure enough for many purposes.

See this page for more information about IPsec.

Some VPN systems use the IPsec protocol, which was standardized by the IETF in 1998 as a relatively low-level adjunct to IP. Others, such as OpenVPN, implement VPN security on top of TCP by using Transport Layer Security (TLS), the successor to the Secure Sockets Layer (SSL). TLS is also on the IETF’s standards track, although it hasn’t yet been fully adopted as of this writing (2017).

A variety of proprietary VPN implementations are also available. These systems generally don’t interoperate with one another or with the standards-based VPN systems, but that’s not necessarily a major drawback if all the endpoints are under your control.

The TLS-based VPN solutions seem to be the marketplace winners at this point. They are just as secure as IPsec and considerably less complicated. Having a free implementation in the form of OpenVPN doesn’t hurt either.

For users at home and at large, a common paradigm is for them to download a small Java or executable component through their web browser. This component then implements VPN connectivity back to the enterprise network. The mechanism is convenient for users, but be aware that the browser-based systems differ widely in their implementations: some offer VPN service through a pseudo-network-interface, while others forward only specific ports. Still others are little more than glorified web proxies.

Be sure you understand the underlying technology of the solutions you’re considering, and don’t expect the impossible. True VPN service (that is, full IP-layer connectivity through a network interface) requires administrative privileges and software installation on the client, whether that client is a Windows system or a Linux laptop. Check browser compatibility too, since the voodoo involved in implementing browser-based VPN solutions often doesn’t translate among browsers.
13.9 Basic network configuration

Only a few steps are involved in adding a new machine to an existing local area network, but every system does it slightly differently. Systems with a GUI installed typically include a control panel for network configuration, but these visual tools address only simple scenarios. On a typical server, you just enter the network configuration directly into text files.

Before bringing up a new machine on a network that is connected to the Internet, secure it (Chapter 27, Security) so that you are not inadvertently inviting attackers onto your local network.

Adding a new machine to a local network goes like this:

1. Assign a unique IP address and hostname.
2. Configure network interfaces and IP addresses.
3. Set up a default route and perhaps fancier routing.
4. Point to a DNS name server to allow access to the rest of the Internet.

If you rely on DHCP for basic provisioning, most of the configuration chores for a new machine are performed on the DHCP server rather than on the new machine itself. New OS installations typically default to configuration through DHCP, so new machines may require no network configuration at all. Refer to the DHCP section starting here for general information.

After any change that might affect startup, always reboot to verify that the machine comes up correctly. Six months later when the power has failed and the machine refuses to boot, it’s hard to remember what change you made that might have caused the problem.

The process of designing and installing a physical network is touched on in Chapter 14, Physical Networking. If you are dealing with an existing network and have a general idea of how it is set up, it may not be necessary for you to read too much more about the physical aspects of networking unless you plan to extend the existing network.

In this section, we review the various issues involved in manual network configuration. This material is general enough to apply to any UNIX or Linux system. In the vendor-specific sections starting here, we address the unique twists that separate the various vendors’ systems.

As you work through basic network configuration on any machine, you’ll find it helpful to test your connectivity with basic tools such as ping and traceroute. See Network troubleshooting for a description of these tools.
Hostname and IP address assignment

See Chapter 16 for more information about DNS.

Administrators have various heartfelt theories about how the mapping from hostnames to IP addresses is best maintained: through the hosts file, LDAP, the DNS system, or perhaps some combination of those options. The conflicting goals are scalability, consistency, and maintainability versus a system that is flexible enough to allow machines to boot and function when not all services are available.

Another consideration when you’re designing your addressing system is the possible need to renumber your hosts in the future. Unless you are using RFC1918 private addresses (see this page), your site’s IP addresses might change when you switch ISPs. Such a transition becomes daunting if you must visit each host on the network to reconfigure its address. To expedite renumbering, you can use hostnames in configuration files and confine address mappings to a few centralized locations such as the DNS database and your DHCP configuration files.

The /etc/hosts file is the oldest and simplest way to map names to IP addresses. Each line starts with an IP address and continues with the various symbolic names by which that address is known.

Here is a typical /etc/hosts file for the host lollipop:

```
127.0.0.1 localhost
::1 localhost ip6-localhost
ff02::1 ip6-allnodes
ff02::2 ip6-allrouters
192.108.21.48 lollipop.atrust.com lollipop loghost
192.108.21.254 chimchim-gw.atrust.com chimchim-gw
192.108.21.1 ns.atrust.com ns
192.225.33.5 licenses.atrust.com license-server
```

A minimalist version would contain only the first three lines. localhost is commonly the first entry in the /etc/hosts file; this entry is unnecessary on many systems, but it doesn’t hurt to include it. You can freely intermix IPv4 and IPv6 addresses.

Because /etc/hosts contains only local mappings and must be maintained on each client system, it’s best reserved for mappings that are needed at boot time (e.g., the host itself, the default gateway, and name servers). Use DNS or LDAP to find mappings for the rest of the local network and the rest of the world. You can also use /etc/hosts to specify mappings that you do not want the rest of the world to know about and therefore do not publish in DNS. (You can also use a split DNS configuration to achieve this goal; see this page.)
The **hostname** command assigns a hostname to a machine. **hostname** is typically run at boot time from one of the startup scripts, which obtains the name to be assigned from a configuration file. (Of course, each system does this slightly differently. See the system-specific sections beginning [here](#) for details.) The hostname should be fully qualified: that is, it should include both the hostname and the DNS domain name, such as anchor.cs.colorado.edu.

At a small site, you can easily dole out hostnames and IP addresses by hand. But when many networks and many different administrative groups are involved, it helps to have some central coordination to ensure uniqueness. For dynamically assigned networking parameters, DHCP takes care of the uniqueness issues.
Network interface and IP configuration

A network interface is a piece of hardware that can potentially be connected to a network. The actual hardware varies widely. It can be an RJ-45 jack with associated signaling hardware for wired Ethernet, a wireless radio, or even a virtual piece of hardware that connects to a virtual network.

Every system has at least two network interfaces: a virtual loopback interface and at least one real network card or port. On PCs with multiple Ethernet jacks, a separate network interface usually controls each jack. (These interfaces quite often have hardware different from that of each other as well.)

On most systems, you can see all the network interfaces with `ip link show` (Linux) or `ifconfig -a` (FreeBSD), whether or not the interfaces have been configured or are currently running. Here’s an example from an Ubuntu system:

```
$ ip link show
1: lo: <LOOPBACK,UP,LOWER_UP> mtu 65536 qdisc noqueue state UNKNOWN
    link/loopback 00:00:00:00:00:00 brd 00:00:00:00:00:00
2: enp0s5: <BROADCAST,MULTICAST,UP,LOWER_UP> mtu 1500 qdisc pfifo_fast
    state UP mode DEFAULT group default qlen 1000
    link/ether 00:1c:42:b4:fa:54 brd ff:ff:ff:ff:ff:ff
```

Interface naming conventions vary. Current versions of Linux try to ensure that interface names don’t change over time, so the names are somewhat arbitrary (e.g., enp0s5). FreeBSD and older Linux kernels use a more traditional driver + instance number scheme, resulting in names like em0 or eth0.

Network hardware often has configurable options that are specific to its media type and have little to do with TCP/IP per se. One common example of this is modern-day Ethernet, wherein an interface card might support 10, 100, 1000, or even 10000 Mb/s in either half-duplex or full-duplex mode. Most equipment defaults to autonegotiation mode, in which both the card and its upstream connection (usually a switch port) try to guess what the other wants to use.

Historically, autonegotiation worked about as well as a blindfolded cowpoke trying to rope a calf. Modern network devices play better together, but autonegotiation is still a possible source of failure. High packet loss rates (especially for large packets) are a common artifact of failed autonegotiation.

If you manually configure a network link, turn off autonegotiation on both sides. It makes intuitive sense that you might be able to manually configure one side of the link and then let the other side automatically adapt to those settings. But alas, that is not how Ethernet autoconfiguration actually works. All participants must agree that the network is either automatically or manually configured.
The exact method by which hardware options like autonegotiation are set varies widely, so we defer discussion of those details to the system-specific sections starting here.

Above the level of interface hardware, every network protocol has its own configuration. IPv4 and IPv6 are the only protocols you might normally want to configure, but it’s important to understand that configurations are defined per interface/protocol pair. In particular, IPv4 and IPv6 are completely separate worlds, each of which has its own configuration.

IP configuration is largely a matter of setting an IP address for the interface. IPv4 also needs to know the subnet mask (“netmask”) for the attached network so that it can distinguish the network and host portions of addresses. At the level of network traffic in and out of an interface, IPv6 does not use netmasks; the network and host portions of an IPv6 address are of fixed size.

In IPv4, you can set the broadcast address to any IP address that’s valid for the network to which the host is attached. Some sites have chosen weird values for the broadcast address in the hope of avoiding certain types of denial-of-service attacks that use broadcast pings, but this is risky and probably overkill. Failure to properly configure every machine’s broadcast address can lead to broadcast storms, in which packets travel from machine to machine until their TTLs expire.

A better way to avoid problems with broadcast pings is to prevent your border routers from forwarding them and to tell individual hosts not to respond to them. IPv6 no longer has broadcasting at all; it’s been replaced with various forms of multicasting.

You can assign more than one IP address to an interface. In the past it was sometimes helpful to do this to allow one machine to host several web sites; however, the need for this feature has been superseded by the HTTP Host header and the SNI feature of TLS. See this page for details.
Routing configuration

This book’s discussion of routing is divided among several sections in this chapter and Chapter 15, IP Routing. Although most of the basic information about routing is found here and in the sections about the ip route (Linux) and route (FreeBSD) commands, you might find it helpful to read the first few sections of Chapter 15 if you need more information.

Routing is performed at the IP layer. When a packet bound for some other host arrives, the packet’s destination IP address is compared with the routes in the kernel’s routing table. If the address matches a route in the table, the packet is forwarded to the next-hop gateway IP address associated with that route.

There are two special cases. First, a packet may be destined for some host on a directly connected network. In this case, the “next-hop gateway” address in the routing table is one of the local host’s own interfaces, and the packet is sent directly to its destination. This type of route is added to the routing table for you by the ifconfig or ip address command when you configure a network interface.

Second, it may be that no route matches the destination address. In this case, the default route is invoked if one exists. Otherwise, an ICMP “network unreachable” or “host unreachable” message is returned to the sender.

Many local area networks have only one way out, so all they need is a single default route that points to the exit. On the Internet backbone, the routers do not have default routes. If there is no routing entry for a destination, that destination cannot be reached.

Each ip route (Linux) or route (FreeBSD) command adds or removes one route. Here are two prototypical commands:

```
linux# ip route add 192.168.45.128/25 via zulu-gw.atrust.net
freebsd# route add -net 192.168.45.128/25 zulu-gw.atrust.net
```

These commands add a route to the 192.168.45.128/25 network through the gateway router zulu-gw.atrust.net, which must be either an adjacent host or one of the local host’s own interfaces. Naturally, the hostname zulu-gw.atrust.net must be resolvable to an IP address. Use a numeric IP address if your DNS server is on the other side of the gateway!

Destination networks were traditionally specified with separate IP addresses and netmasks, but all routing-related commands now understand CIDR notation (e.g., 128.138.176.0/20). CIDR notation is clearer and relieves you of the need to fuss over some of the system-specific syntax issues.

Some other tricks:

- To inspect existing routes, use the command netstat -nr, or netstat -r if you want to see names instead of numbers. Numbers are often better if you are debugging, since
the name lookup may be the thing that is broken. An example of netstat output is shown here. On Linux, ip route show is the officially blessed command for seeing routes. However, we find its output less clear than netstat’s.

- Use the keyword default instead of an address or network name to set the system’s default route. This mnemonic is identical to 0.0.0.0/0, which matches any address and is less specific than any real routing destination.

- Use ip route del (Linux) or route del (FreeBSD) to remove entries from the routing table.

- Run ip route flush (Linux) or route flush (FreeBSD) to initialize the routing table and start over.

- IPv6 routes are set up similarly to IPv4 routes; include the -6 option to route to tell it that you’re setting an IPv6 routing entry. ip route can normally recognize IPv6 routes on its own (by inspecting the format of the addresses), but it accepts the -6 argument, too.

- /etc/networks maps names to network numbers, much like the hosts file maps hostnames to IP addresses. Commands such as ip and route that expect a network number can accept a name if it is listed in the networks file. Network names can also be listed in DNS; see RFC1101.
DNS configuration

To configure a machine as a DNS client, you need only set up the `/etc/resolv.conf` file. DNS service is not strictly required, but it’s hard to imagine a situation in which you’d want to eliminate it completely.

The `resolv.conf` file lists the DNS domains that should be searched to resolve names that are incomplete (that is, not fully qualified, such as anchor instead of anchor.cs.colorado.edu) and the IP addresses of the name servers to contact for name lookups. A sample is shown below; for more details, see [this page](#).

```
search cs.colorado.edu colorado.edu
nameserver 128.138.242.1
nameserver 128.138.243.151
nameserver 192.108.21.1
```

`/etc/resolv.conf` should list the “closest” stable name server first. Servers are contacted in order, and the timeout after which the next server in line is tried can be quite long. You can have up to three `nameserver` entries. If possible, you should always have more than one.

If the local host obtains the addresses of its DNS servers through DHCP, the DHCP client software stuffs these addresses into the `resolv.conf` file for you when it obtains the leases. Since DHCP configuration is the default for most systems, you generally need not configure the `resolv.conf` file manually if your DHCP server has been set up correctly.

Many sites use Microsoft’s Active Directory DNS server implementation. That works fine with the standard UNIX and Linux `resolv.conf`; there’s no need to do anything differently.
System-specific network configuration

On early UNIX systems, you configured the network by editing the system startup scripts and directly changing the commands they contained. Modern systems have read-only scripts; they cover a variety of configuration scenarios and choose among them by reusing information from other system files or consulting configuration files of their own.

Although this separation of configuration and implementation is a good idea, every system does it a little bit differently. The format and use of the /etc/hosts and /etc/resolv.conf files are relatively consistent among UNIX and Linux systems, but that’s about all you can count on for sure.

Most systems offer some sort of GUI interface for basic configuration tasks, but the mapping between the visual interface and the configuration files behind the scenes is often unclear. In addition, the GUIs tend to ignore advanced configurations, and they are relatively inconvenient for remote and automated administration.

In the next sections, we pick apart some of the variations among our example systems, describe what’s going on under the hood, and cover the details of network configuration for each of our supported operating systems. In particular, we cover

- Basic configuration
- DHCP client configuration
- Dynamic reconfiguration and tuning
- Security, firewalls, filtering, and NAT configuration

Keep in mind that most network configuration happens at boot time, so there’s some overlap between the information here and the information presented in Chapter 2, Booting and System Management Daemons.
13.10 Linux Networking

Linux developers love to tinker, and they often implement features and algorithms that aren’t yet accepted standards. One example is the Linux kernel’s addition of pluggable congestion-control algorithms in release 2.6.13. The several options include variations for lossy networks, high-speed WANs with lots of packet loss, satellite links, and more. The standard TCP “reno” mechanism (slow start, congestion avoidance, fast retransmit, and fast recovery) is still used by default. A variant might be more appropriate for your environment (but probably not). See [lwn.net/Articles/701165](http://lwn.net/Articles/701165) for some hints on when to consider the use of alternate congestion control algorithms.

After any change to a file that controls network configuration at boot time, you may need either to reboot or to bring the network interface down and then up again for your change to take effect. You can use `ifdown interface` and `ifup interface` for this purpose on most Linux systems.
NetworkManager

Linux support for mobile networking was relatively scattershot until the advent of NetworkManager in 2004. It consists of a service that’s run continuously, along with a system tray app for configuring individual network interfaces. In addition to various kinds of wired network, NetworkManager also handles transient wireless networks, wireless broadband, and VPNs. It continually assesses the available networks and shifts service to “preferred” networks as they become available. Wired networks are most preferred, followed by familiar wireless networks.

This system represented quite a change for Linux network configuration. In addition to being more fluid than the traditional static configuration, it’s also designed to be run and managed by users rather than system administrators. NetworkManager has been widely adopted by Linux distributions, including all our examples, but in an effort to avoid breaking existing scripts and setups, it’s usually made available as a sort of “parallel universe” of network configuration in addition to whatever traditional network configuration was used in the past.

Debian and Ubuntu run NetworkManager by default, but keep the statically configured network interfaces out of the NetworkManager domain. Red Hat and CentOS don’t run NetworkManager by default at all.

NetworkManager is primarily of use on laptops, since their network environment may change frequently. For servers and desktop systems, NetworkManager isn’t necessary and may in fact complicate administration. In these environments, it should be ignored or configured out.
**ip: manually configure a network**

Linux systems formerly used the same basic commands for network configuration and status checking as traditional UNIX: `ifconfig`, `route`, and `netstat`. These are still available on most distributions, but active development has moved on to the `iproute2` package, which features the commands `ip` (for most everyday network configuration, including routing) and `ss` (for examining the state of network sockets, roughly replacing `netstat`).

If you’re accustomed to the traditional commands, it’s worth the effort to transition your brain to `ip`. The legacy commands won’t be around forever, and although they cover the common configuration scenarios, they don’t give access to the full feature set of the Linux networking stack. `ip` is cleaner and more regular.

`ip` takes a second argument for you to specify what kind of object you want to configure or examine. There are many options, but the common ones are `ip link` for configuring network interfaces, `ip address` for binding network addresses to interfaces, and `ip route` for changing or printing the routing table. You can abbreviate `ip` arguments, so `ip ad` is the same as `ip address`. We show full names for clarity.

Most objects understand `list` or `show` to print out a summary of their current status, so `ip link show` prints a list of network interfaces, `ip route show` dumps the current routing table, and `ip address show` lists all assigned IP addresses.

The man pages for `ip` are divided by subcommand. For example, to see detailed information about interface configuration, run `man ip-link`. You can also run `ip link help` to see a short cheat sheet.

The UNIX `ifconfig` command conflates the concept of interface configuration with the concept of configuring the settings for a particular network protocol. In fact, several protocols can run on a given network interface (the prime example being simultaneous IPv4 and IPv6), and each of those protocols can support multiple addresses, so `ip`’s distinction between `ip link` and `ip address` is actually quite smart. Most of what system administrators traditionally think of as “interface configuration” really has to do with setting up IPv4 and IPv6.

`ip` accepts a `-4` or `-6` argument to target IPv4 or IPv6 explicitly, but it’s rarely necessary to specify these options. `ip` guesses the right mode just by looking at the format of the addresses you provide.

Basic configuration of an interface looks like this:

```
# ip address add 192.168.1.13/26 broadcast 192.168.1.63 dev enp0s5
```

In this case the `broadcast` clause is superfluous because that would be the default value anyway, given the netmask. But this is how you would set it if you needed to.
Of course, in daily life you won’t normally be setting up network addresses by hand. The next sections describe how our example distributions handle static configuration of the network from the perspective of configuration files.
Debian and Ubuntu network configuration

As shown in Table 13.6, Debian and Ubuntu configure the network in /etc/hostname and /etc/network/interfaces, with a bit of help from the file /etc/network/options.

Table 13.6: Ubuntu network configuration files in /etc

<table>
<thead>
<tr>
<th>File</th>
<th>What’s set there</th>
</tr>
</thead>
<tbody>
<tr>
<td>hostname</td>
<td>Hostname</td>
</tr>
<tr>
<td>network/interfaces</td>
<td>IP address, netmask, default route</td>
</tr>
</tbody>
</table>

The hostname is set in /etc/hostname. The name in this file should be fully qualified; its value is used in a variety of contexts, some of which require qualification.

The IP address, netmask, and default gateway are set in /etc/network/interfaces. A line starting with the iface keyword introduces each interface. The iface line can be followed by indented lines that specify additional parameters. For example,

```
auto lo enp0s5
iface lo inet loopback
iface enp0s5 inet static
    address 192.168.1.102
    netmask 255.255.255.0
    gateway 192.168.1.254
```

The ifup and ifdown commands read this file and bring the interfaces up or down by calling lower-level commands (such as ip) with the appropriate parameters. The auto clause specifies the interfaces to be brought up at boot time or whenever ifup -a is run.

The inet keyword in the iface line is the address family, IPv4. To configure IPv6 as well, include an inet6 configuration.

The keyword static is called a “method” and specifies that the IP address and netmask for enp0s5 are directly assigned. The address and netmask lines are required for static configurations. The gateway line specifies the address of the default network gateway and is used to install a default route.

To configure interfaces with DHCP, just specify that in the interfaces file:

```
iface enp0s5 inet dhcp
```
Red Hat and CentOS network configuration

Red Hat and CentOS’s network configuration revolves around /etc/sysconfig. Table 13.7 shows the various configuration files.

Table 13.7: Red Hat network configuration files in /etc/sysconfig

<table>
<thead>
<tr>
<th>File</th>
<th>What’s set there</th>
</tr>
</thead>
<tbody>
<tr>
<td>network</td>
<td>Hostname, default route</td>
</tr>
<tr>
<td>network-scripts/ifcfg-ifname</td>
<td>Per-interface parameters: IP address, netmask, etc.</td>
</tr>
<tr>
<td>network-scripts/route-ifname</td>
<td>Per-interface routing: arguments to ip route</td>
</tr>
</tbody>
</table>

You set the machine’s hostname in /etc/sysconfig/network, which also contains lines that specify the machine’s DNS domain and default gateway. (Essentially, this file is where you specify all interface-independent network settings.)

For example, here is a network file for a host with a single Ethernet interface:

```
NETWORKING=yes
NETWORKING_IPV6=no
HOSTNAME=redhat.toaddranch.com
DOMAINNAME=toaddranch.com ## optional
GATEWAY=192.168.1.254
```

Interface-specific data is stored in /etc/sysconfig/network-scripts/ifcfg-ifname, where ifname is the name of the network interface. These configuration files set the IP address, netmask, network, and broadcast address for each interface. They also include a line that specifies whether the interface should be configured “up” at boot time.

A generic machine has files for an Ethernet interface (eth0) and for the loopback interface (lo). For example,

```
DEVICE=eth0
IPADDR=192.168.1.13
NETMASK=255.255.255.0
NETWORK=192.168.1.0
BROADCAST=192.168.1.255
MTU=1500
ONBOOT=yes
```

and
are the ifcfg-eth0 and ifcfg-lo files for the machine redhat.toadranch.com described in the network file above.

A DHCP-based setup for eth0 is even simpler:

```
DEVICE=eth0
BOOTPROTO= dhcp
ONBOOT=yes
```

After changing configuration information in /etc/sysconfig, run `ifdown ifname` followed by `ifup ifname` for the appropriate interface. If you reconfigure multiple interfaces at once, you can use the command `sysctl restart network` to reset networking.

Lines in network-scripts/route-ifname are passed as arguments to `ip route` when the corresponding interface is configured. For example, the line

```
default via 192.168.0.1
```

sets a default route. This isn’t really an interface-specific configuration, but for clarity, it should go in the file that corresponds to the interface on which you expect default-routed packets to be transmitted.
Linux network hardware options

The `ethtool` command queries and sets a network interface’s media-specific parameters such as link speed and duplex. It replaces the old `mii-tool` command, but some systems still include both. If `ethtool` is not installed by default, it’s usually included in an optional package of its own (also called `ethtool`).

You can query the status of an interface just by naming it. For example, this eth0 interface (a generic NIC on a PC motherboard) has autonegotiation enabled and is currently running at full speed:

```
# ethtool eth0
Settings for eth0:
   Supported ports: [ TP MII ]
   Supported link modes:  10baseT/Half  10baseT/Full
                          100baseT/Half 100baseT/Full
                          1000baseT/Half 1000baseT/Full
   Supports auto-negotiation: Yes
   Advertised link modes:  10baseT/Half  10baseT/Full
                           100baseT/Half 100baseT/Full
                           1000baseT/Half 1000baseT/Full
   Advertised auto-negotiation: Yes
   Speed: 1000Mb/s
   Duplex: Full
   Port: MII
   PHYAD: 0
   Transceiver: internal
   Auto-negotiation: on
   Supports Wake-on: pumbg
   Wake-on: g
   Current message level: 0x00000033 (51)
   Link detected: yes
```

To lock this interface to 100 Mb/s full duplex, use the command

```
# ethtool -s eth0 speed 100 duplex full
```

If you are trying to determine whether autonegotiation is reliable in your environment, you may also find `ethtool -r` helpful. It forces the parameters of the link to be renegotiated immediately.

Another useful option is `-k`, which shows what protocol-related tasks are assigned to the network interface rather than being performed by the kernel. Most interfaces can calculate checksums, and some can assist with segmentation as well. Unless you believe that a network interface is not doing these tasks reliably, it’s always better to offload them. You can use `ethtool -K` in combination with various suboptions to force or disable specific types of offloading. (The `-k` option shows current values and the `-K` option sets them.)
Any changes you make with `ethtool` are transient. If you want them to be enforced consistently, make sure that `ethtool` gets run as part of the system’s network configuration. It’s best to do this as part of the per-interface configuration; if you just arrange to have some `ethtool` commands run at boot time, your configuration will not properly cover cases in which the interfaces are restarted without a reboot of the system.

On Red Hat and CentOS systems, you can include an `ETHTOOL_OPTS=` line in the configuration file for the interface underneath `/etc/sysconfig/network-scripts`. The `ifup` command passes the entire line as arguments to `ethtool`.

In Debian and Ubuntu, you can run `ethtool` commands directly from the configuration for a particular network in `/etc/network/interfaces`.
**Linux TCP/IP options**

Linux puts a representation of each tunable kernel variable into the `/proc` virtual filesystem. See [Tuning Linux kernel parameters](#) for general information about the `/proc` mechanism.

The networking variables are under `/proc/sys/net/ipv4` and `/proc/sys/net/ipv6`. We formerly showed a complete list here, but there are too many to list these days.

The `ipv4` directory includes a lot more parameters than does the `ipv6` directory, but that’s mostly because IP-version-independent protocols such as TCP and UDP confine their parameters to the `ipv4` directory. A prefix such as `tcp_` or `udp_` tells you which protocol the parameter relates to.

The `conf` subdirectories within `ipv4` and `ipv6` contain parameters that are set per interface. They include subdirectories `all` and `default` and a subdirectory for each interface (including the loopback). Each subdirectory has the same set of files.

```
ubuntu$ ls -F /proc/sys/net/ipv4/conf/default
accept_local drop_gratuitous_arp proxy_arp
accept_redirects drop_unicast_in_12_multicast proxy_arp_pvlan
accept_source_route force_igmp_version route_localnet
arp_accept forwarding rp_filter
arp_announce igmpv2_unsolicited_report_interval secure_redirects
arp_filter igmpv3_unsolicited_report_interval send_redirects
arp_ignore ignore_routes_with_linkdown shared_media
arp_notify log_martians src_valid_mark
tcp_bootp Relay mc_forwarding tag
disable_policy medium_id
disable_xfrm promote_secondaries
```

If you change a variable in the `conf/enp0s5` subdirectory, for example, your change applies to that interface only. If you change the value in the `conf/all` directory, you might expect it to set the corresponding value for all existing interfaces, but this is not what happens. Each variable has its own rules for accepting changes via `all`. Some values are ORed with the current values, some are ANDEd, and still others are MAXed or MINed. As far as we are aware, there is no documentation for this process except in the kernel source code, so the whole debacle is probably best avoided. Just confine your modifications to individual interfaces.

If you change a variable in the `conf/default` directory, the new value propagates to any interfaces that are later configured. On the other hand, it’s nice to keep the defaults unmolested as reference information; they make a nice sanity check if you want to undo other changes.

The `/proc/sys/net/ipv4/neigh` and `/proc/sys/net/ipv6/neigh` directories also contain a subdirectory for each interface. The files in each subdirectory control ARP table management and IPv6 neighbor discovery for that interface. Here is the list of variables; the ones starting with `gc` (for garbage collection) determine how ARP table entries are timed out and discarded.
To see the value of a variable, use `cat`. To set it, you can use `echo` redirected to the proper filename, but the `sysctl` command (which is just a command interface to the same variables) is often easier.

For example, the command

```
ubuntu$ cat /proc/sys/net/ipv4/icmp_echo_ignore_broadcasts
0
```

shows that this variable’s value is 0, meaning that broadcast pings are not ignored. To set it to 1 (and avoid falling prey to Smurf-type denial of service attacks), run

```
ubuntu$ sudo sh -c "echo 1 > icmp_echo_ignore_broadcasts"
```

from the `/proc/sys/net` directory. (If you try this command in the form `sudo echo 1 > icmp_echo_ignore_broadcasts`, you just generate a “permission denied” message—your shell attempts to open the output file before it runs `sudo`. You want the `sudo` to apply to both the `echo` command and the redirection. Ergo, you must create a root subshell in which to execute the entire command.)

You can also use the `sysctl` command to achieve the same configuration:

```
ubuntu$ sysctl net.ipv4.icmp_echo_ignore_broadcasts=1
```

`sysctl` variable names are pathnames relative to `/proc/sys`. Dots are the traditional separator, but `sysctl` also accepts slashes if you prefer them.

You are typically logged in over the same network you are tweaking as you adjust these variables, so be careful! You can mess things up badly enough to require a reboot from the console to recover, which might be inconvenient if the system happens to be in Point Barrow, Alaska, and it’s January. Test-tune these variables on your desktop system before you even think of tweaking a production machine.

To change any of these parameters permanently (or more accurately, to reset them every time the system boots), add the appropriate variables to `/etc/sysctl.conf`, which is read by the `sysctl` command at boot time. For example, the line

```
net.ipv4.ip_forward=0
```

in the `/etc/sysctl.conf` file turns off IP forwarding on this host.
Some of the options under /proc are better documented than others. Your best bet is to look at the man page for the protocol in question in section 7 of the manuals. For example, `man 7 icmp` documents six of the eight available options. (You must have man pages for the Linux kernel installed to see man pages about protocols.)

You can also look at the `ip-sysctl.txt` file in the kernel source distribution for some good comments. If you don’t have kernel source installed, just Google for ip-sysctl-txt to reach the same document.
Security-related kernel variables

Table 13.8 shows Linux’s default behavior with regard to various touchy network issues. For a brief description of the implications of these behaviors, see this page. We recommend that you verify the values of these variables so that you do not answer broadcast pings, do not listen to routing redirects, and do not accept source-routed packets. These should be the defaults on current distributions except for `accept_redirects`.

Table 13.8: Default security-related network behaviors in Linux

<table>
<thead>
<tr>
<th>Feature</th>
<th>Host</th>
<th>Gateway</th>
<th>Control file (in /proc/sys/net/ipv4)</th>
</tr>
</thead>
<tbody>
<tr>
<td>IP forwarding</td>
<td>off</td>
<td>on</td>
<td><code>ip_forward</code> for the whole system</td>
</tr>
<tr>
<td></td>
<td></td>
<td></td>
<td><code>conf/interface/forwarding</code> per interface&lt;sup&gt;a&lt;/sup&gt;</td>
</tr>
<tr>
<td>ICMP redirects</td>
<td>obeys</td>
<td>ignores</td>
<td><code>conf/interface/accept_redirects</code></td>
</tr>
<tr>
<td>Source routing</td>
<td>off</td>
<td>off</td>
<td><code>conf/interface/accept_source_route</code></td>
</tr>
<tr>
<td>Broadcast ping</td>
<td>ignores</td>
<td>ignores</td>
<td><code>icmp_echo_ignore_broadcasts</code></td>
</tr>
</tbody>
</table>

<sup>a</sup> The *interface* can be either a specific interface name or **all**.
13.11 FreeBSD networking

As a direct descendant of the BSD lineage, FreeBSD remains something of a reference implementation of TCP/IP. It lacks many of the elaborations that complicate the Linux networking stack. From the standpoint of system administrators, FreeBSD network configuration is simple and direct.
**ifconfig: configure network interfaces**

`ifconfig` enables or disables a network interface, sets its IP address and subnet mask, and sets various other options and parameters. It is usually run at boot time with command-line parameters taken from config files, but you can also run it by hand to make changes on the fly. Be careful if you are making `ifconfig` changes and are logged in remotely—many a sysadmin has been locked out this way and had to drive in to fix things.

An `ifconfig` command most commonly has the form

```
ifconfig interface [family] address options ...
```

For example, the command

```
ifconfig em0 192.168.1.13/26 up
```

sets the IPv4 address and netmask associated with the interface em0 and readies the interface for use.

`interface` identifies the hardware interface to which the command applies. The loopback interface is named lo0. The names of real interfaces vary according to their hardware drivers. `ifconfig -a` lists the system’s network interfaces and summarizes their current settings.

The `family` parameter tells `ifconfig` which network protocol (“address family”) you want to configure. You can set up multiple protocols on an interface and use them all simultaneously, but they must be configured separately. The main options here are `inet` for IPv4 and `inet6` for IPv6; `inet` is assumed if you omit the parameter.

The `address` parameter specifies the interface’s IP address. A hostname is also acceptable here, but the hostname must be resolvable to an IP address at boot time. For a machine’s primary interface, this means that the hostname must appear in the local `hosts` file, since other name resolution methods depend on the network having been initialized.

The keyword `up` turns the interface on; `down` turns it off. When an `ifconfig` command assigns an IP address to an interface, as in the example above, the `up` parameter is implicit and need not be mentioned by name.

For subnetted networks, you can specify a CIDR-style netmask as shown in the example above, or you can include a separate `netmask` argument. The mask can be specified in dotted decimal notation or as a 4-byte hexadecimal number beginning with `0x`.

The `broadcast` option specifies the IP broadcast address for the interface, expressed in either hex or dotted quad notation. The default broadcast address is one in which the host part is set to all 1s. In the `ifconfig` example above, the autoconfigured broadcast address is 192.168.1.61.
FreeBSD network hardware configuration

FreeBSD does not have a dedicated command analogous to Linux’s `ethertool`. Instead, `ifconfig` passes configuration information down to the network interface driver through the `media` and `mediaopt` clauses. The legal values for these options vary with the hardware. To find the list, read the man page for the specific driver.

For example, an interface named `em0` uses the “em” driver. `man 4 em` shows that this is the driver for certain types of Intel-based wired Ethernet hardware. To force this interface to gigabit mode using four-pair cabling (the typical configuration), the command would be

```
# ifconfig em0 media 1000baseT mediaopt full-duplex
```

You can include these media options along with other configuration clauses for the interface.
FreeBSD boot-time network configuration

FreeBSD’s static configuration system is mercifully simple. All the network parameters live in /etc/rc.conf, along with other system-wide settings. Here’s a typical configuration:

```bash
hostname="freebeer"
ifconfig_em0="inet 192.168.0.48 netmask 255.255.255.0"
defaultrouter="192.168.0.1"
```

Each network interface has its own ifconfig_* variable. The value of the variable is simply passed to ifconfig as a series of command-line arguments. The defaultrouter clause identifies a default network gateway.

To obtain the system’s networking configuration from a DHCP server, use the following token:

```bash
ifconfig_em0="DHCP"
```

This form is magic and is not passed on to ifconfig, which wouldn’t know how to interpret a DHCP argument. Instead, it makes the startup scripts run the command dhclient em0. To modify the operational parameters of the DHCP system (timeouts and such), set them in /etc/dhcclient.conf. The default version of this file is empty except for comments, and you shouldn’t normally need to modify it.

If you modify the network configuration, you can run service netif restart to repeat the initial configuration procedure. If you changed the defaultrouter parameter, also run service routing restart.
FreeBSD TCP/IP configuration

FreeBSD’s kernel-level networking options are controlled similarly to those of Linux (see this page), except that there’s no /proc hierarchy you can go rooting around in. Instead, run `sysctl -ad` to list the available parameters and their one-line descriptions. There are a lot of them (5,495 on FreeBSD 11), so you need to grep for likely suspects such as “redirect” or “^net”.

Table 13.9 lists a selection of security-related parameters.

Table 13.9: Default security-related network parameters in FreeBSD

<table>
<thead>
<tr>
<th>Parameter</th>
<th>Dfl</th>
<th>What it does when set to 1</th>
</tr>
</thead>
<tbody>
<tr>
<td>net.inet.ip.forwarding</td>
<td>0</td>
<td>Acts as a router for IPv4 packets</td>
</tr>
<tr>
<td>net.inet6.ip6.forwarding</td>
<td>0</td>
<td>Acts as a router for IPv6 packets</td>
</tr>
<tr>
<td>net.inet.tcp.blackhole</td>
<td>0</td>
<td>Disables “unreachable” messages for closed ports</td>
</tr>
<tr>
<td>net.inet.udp.blackhole</td>
<td>0</td>
<td>Does not send RST packets for closed TCP ports</td>
</tr>
<tr>
<td>net.inet.icmp.drop_redirect</td>
<td>0</td>
<td>Ignores IPv4 ICMP redirects</td>
</tr>
<tr>
<td>net.inet6.icmp6.redirectaccept</td>
<td>1</td>
<td>Accepts (obeys) IPv6 ICMP redirects</td>
</tr>
<tr>
<td>net.inet.ip.accept_sourceroute</td>
<td>0</td>
<td>Allows source-routed IPv4 packets</td>
</tr>
</tbody>
</table>

The blackhole options are potentially useful on systems that you want to shield from port scanners, but they do change the standard behaviors of UDP and TCP. You might also want to disable acceptance of ICMP redirects for both IPv4 and IPv6.

You can set parameters in the running kernel with `sysctl`. For example,

```
$ sudo sysctl net.inet.icmp.drop_redirect=1
```

To have the parameter set at boot time, list it in `/etc/sysctl.conf`.

```
net.inet.icmp.drop_redirect=1
```
13.12 Network Troubleshooting

Several good tools are available for debugging a network at the TCP/IP layer. Most give low-level information, so you must understand the main ideas of TCP/IP and routing to use them.

In this section, we start with some general troubleshooting strategy. We then cover several essential tools, including ping, traceroute, tcpdump, and Wireshark. We don’t discuss the arp, ndp, ss, or netstat commands in this chapter, though they, too, are useful debugging tools.

Before you attack your network, consider these principles:

- Make one change at a time. Test each change to make sure that it had the effect you intended. Back out any changes that have an undesired effect.
- Document the situation as it was before you got involved, and document every change you make along the way.
- Start at one end of a system or network and work through the system’s critical components until you reach the problem. For example, you might start by looking at the network configuration on a client, work your way up to the physical connections, investigate the network hardware, and finally, check the server’s physical connections and software configuration.
- Or, use the layers of the network to negotiate the problem. Start at the “top” or “bottom” and work your way through the protocol stack.

This last point deserves a bit more discussion. As described here, the architecture of TCP/IP defines several layers of abstraction at which components of the network can function. For example, HTTP depends on TCP, TCP depends on IP, IP depends on the Ethernet protocol, and the Ethernet protocol depends on the integrity of the network cable. You can dramatically reduce the amount of time spent debugging a problem if you first figure out which layer is misbehaving.

Ask yourself questions like these as you work up or down the stack:

- Do you have physical connectivity and a link light?
- Is your interface configured properly?
- Do your ARP tables show other hosts?
- Is there a firewall on your local machine?
- Is there a firewall anywhere between you and the destination?
- If firewalls are involved, do they pass ICMP ping packets and responses?
- Can you ping the localhost address (127.0.0.1)?
- Can you ping other local hosts by IP address?
- Is DNS working properly?
- Can you ping other local hosts by hostname?
- Can you ping hosts on another network?
- Do high-level services such as web and SSH servers work?
- Did you really check the firewalls?

If a machine hangs at boot time, boots very slowly, or hangs on inbound SSH connections, DNS should be a prime suspect. Most systems use an approach to name resolution that’s configurable in `/etc/nsswitch.conf`. If the system runs `nscl`, the name service caching daemon, that component deserves some suspicion as well. If `nscl` crashes or is misconfigured, name lookups are affected. Use the `getent` command to check whether your resolver and name servers are working properly (e.g., `getent hosts google.com`).

Once you’ve identified where the problem lies and have a fix in mind, step back to consider the effect that your subsequent tests and prospective fixes will have on other services and hosts.
ping: check to see if a host is alive

The **ping** command and its IPv6 twin **ping6** are embarrassingly simple, but in many situations they are the only commands you need for network debugging. They send an ICMP ECHO_REQUEST packet to a target host and wait to see if the host answers back.

You can use **ping** to check the status of individual hosts and to test segments of the network. Routing tables, physical networks, and gateways are all involved in processing a ping, so the network must be more or less working for **ping** to succeed. If **ping** doesn’t work, you can be pretty sure that nothing more sophisticated will work either.

However, this rule does not apply to networks or hosts that block ICMP echo requests with a firewall. (Recent versions of Windows block ping requests by default.) Make sure that a firewall isn’t interfering with your debugging before you conclude that the target host is ignoring a ping. You might consider disabling a meddlesome firewall for a short period of time to facilitate debugging.

If your network is in bad shape, chances are that DNS is not working. Simplify the situation by using numeric IP addresses when pinging, and use **ping**’s `-n` option to prevent **ping** from attempting to do reverse lookups on IP addresses—these lookups also trigger DNS requests.

Be aware of the firewall issue if you’re using **ping** to check your Internet connectivity, too. Some well-known sites answer **ping** packets and others don’t. We’ve found google.com to be a consistent responder.

Most versions of **ping** run in an infinite loop unless you supply a packet count argument. Once you’ve had your fill of pinging, type the interrupt character (usually `<Control-C>`) to get out.

Here’s an example:

```
linux$ ping beast
PING beast (10.1.1.46): 56 bytes of data.
64 bytes from beast (10.1.1.46): icmp_seq=0 ttl=54 time=48.3ms
64 bytes from beast (10.1.1.46): icmp_seq=1 ttl=54 time=46.4ms
64 bytes from beast (10.1.1.46): icmp_seq=2 ttl=54 time=88.7ms
^C
--- beast ping statistics ---
3 packets transmitted, 3 received, 0% packet loss, time 2026ms
rtt min/avg/max/mdev = 46.490/61.202/88.731/19.481 ms
```

The output for beast shows the host’s IP address, the ICMP sequence number of each response packet, and the round trip travel time. The most obvious thing that the output above tells you is that the server beast is alive and connected to the network.

The ICMP sequence number is a particularly valuable piece of information. Discontinuities in the sequence indicate dropped packets. They’re normally accompanied by a message for each
missing packet.

Despite the fact that IP does not guarantee the delivery of packets, a healthy network should drop very few of them. Lost-packet problems are important to track down because they tend to be masked by higher-level protocols. The network may appear to function correctly, but it will be slower than it ought to be, not only because of the retransmitted packets but also because of the protocol overhead needed to detect and manage them.

To track down the cause of disappearing packets, first run traceroute (see the next section) to discover the route that packets are taking to the target host. Then ping the intermediate gateways in sequence to discover which link is dropping packets. To pin down the problem, you need to send a fair number of packets. The fault generally lies on the link between the last gateway you can ping without loss of packets and the gateway beyond that.

The round trip time reported by ping can afford insight into the overall performance of a path through a network. Moderate variations in round trip time do not usually indicate problems. Packets may occasionally be delayed by tens or hundreds of milliseconds for no apparent reason; that’s just the way IP works. You should see a fairly consistent round trip time for the majority of packets, with occasional lapses. Many of today’s routers implement rate-limited or low-priority responses to ICMP packets, which means that a router may defer responding to your ping if it is already dealing with a lot of other traffic.

The ping program can send echo request packets of any size, so by using a packet larger than the MTU of the network (1,500 bytes for Ethernet), you can force fragmentation. This practice helps you identify media errors or other low-level issues such as problems with a congested network or VPN. You specify the desired packet size in bytes with the -s flag:

$ ping -s 1500 cuinfo.cornell.edu

Note that even a simple command like ping can have dramatic effects. In 1998, the so-called Ping of Death attack crashed large numbers of UNIX and Windows systems. It was launched simply by transmission of an overly large ping packet. When the fragmented packet was reassembled, it filled the receiver’s memory buffer and crashed the machine. The Ping of Death issue has long since been fixed, but keep in mind several other caveats regarding ping.

First, it’s hard to distinguish the failure of a network from the failure of a server with only the ping command. In an environment where ping tests normally work, a failed ping just tells you that something is wrong.

It’s also worth noting that a successful ping does not guarantee much about the target machine’s state. Echo request packets are handled within the IP protocol stack and do not require a server process to be running on the probed host. A response guarantees only that a machine is powered on and has not experienced a kernel panic. You’ll need higher-level methods to verify the availability of individual services such as HTTP and DNS.
**traceroute: trace IP packets**

*traceroute*, originally written by Van Jacobson, uncovers the sequence of gateways through which an IP packet travels to reach its destination. All modern operating systems come with some version of *traceroute*. Even Windows has it, but the command is spelled *tracert* (extra history points if you can guess why).

The syntax is simply

```
traceroute hostname
```

Most of the variety of options are not important in daily use. As usual, the *hostname* can be specified as either a DNS name or an IP address. The output is simply a list of hosts, starting with the first gateway and ending at the destination. For example, a *traceroute* from our host jaguar to our host nubark produces the following output:

```
$ traceroute nubark
traceroute to nubark (192.168.2.10), 30 hops max, 38 byte packets
  1 lab-gw (172.16.8.254) 0.840 ms 0.693 ms 0.671 ms
  2 dmz-gw (192.168.1.254) 4.642 ms 4.582 ms 4.674 ms
  3 nubark (192.168.2.10) 7.959 ms 5.949 ms 5.908 ms
```

From this output we can tell that jaguar is three hops away from nubark, and we can see which gateways are involved in the connection. The round trip time for each gateway is also shown—three samples for each hop are measured and displayed. A typical *traceroute* between Internet hosts often includes more than 15 hops, even if the two sites are just across town.

*traceroute* works by setting the time-to-live field (TTL, actually “hop count to live”) of an outbound packet to an artificially low number. As packets arrive at a gateway, their TTL is decreased. When a gateway decreases the TTL to 0, it discards the packet and sends an ICMP “time exceeded” message back to the originating host.

See [this page](#) for more information about reverse DNS lookups.

The first three *traceroute* packets above have their TTL set to 1. The first gateway to see such a packet (lab-gw in this case) determines that the TTL has been exceeded and notifies jaguar of the dropped packet by sending back an ICMP message. The sender’s IP address in the header of the error packet identifies the gateway, and *traceroute* looks up this address in DNS to find the gateway’s hostname.

To identify the second-hop gateway, *traceroute* sends out a second round of packets with TTL fields set to 2. The first gateway routes the packets and decreases their TTL by 1. At the second gateway, the packets are then dropped and ICMP error messages are generated as before. This
process continues until the TTL is equal to the number of hops to the destination host and the packets reach their destination successfully.

Most routers send their ICMP messages from the interface “closest” to the destination. If you run `traceroute` backward from the destination host, you may see different IP addresses being used to identify the same set of routers. You might also discover that packets flowing in the reverse direction take a completely different path, a configuration known as asymmetric routing.

Since `traceroute` sends three packets for each value of the TTL field, you may sometimes observe an interesting artifact. If an intervening gateway multiplexes traffic across several routes, the packets might be returned by different hosts; in this case, `traceroute` simply prints them all.

Let’s look at a more interesting example from a host in Switzerland to caida.org at the San Diego Supercomputer Center:

```
linux$ traceroute caida.org
traceroute to caida.org (192.172.226.78), 30 hops max, 46 byte packets
1  gw-oetiker.init7.net (213.144.138.193) 1.122 ms 0.182 ms 0.170 ms
2  r1zur1.core.init7.net (77.109.128.209) 0.527 ms 0.204 ms 0.202 ms
3  r1fra1.core.init7.net (77.109.128.250) 18.27 ms 6.99 ms 16.59 ms
4  r1ams1.core.init7.net (77.109.128.154) 19.54 ms 21.85 ms 13.51 ms
5  r1lon1.core.init7.net (77.109.128.150) 19.16 ms 21.15 ms 24.86 ms
6  r1lax1.ce.init7.net (82.197.168.69) 158.23 ms 158.22 ms 158.27 ms
7  cenic.laap.net (198.32.146.32) 158.34 ms 158.30 ms 158.24 ms
8  dc-lax-core2-ge.cenic.net (137.164.46.119) 158.60 ms * 158.71 ms
9  dc-tus-aggl-core2-10ge.cenic.net (137.164.46.7) 159 ms 159 ms 159 ms
10 dc-sdsc2-tus-dc-ge.cenic.net (137.164.24.174) 161 ms 161 ms 161 ms
11 pinot.sdsc.edu (198.17.46.56) 161.559 ms 161.381 ms 161.439 ms
12 rommie.caida.org (192.172.226.78) 161.442 ms 161.445 ms 161.532 ms
```

This output shows that packets travel inside Init Seven’s network for a long time. Sometimes we can guess the location of the gateways from their names. Init Seven’s core stretches all the way from Zurich (zur) to Frankfurt (fra), Amsterdam (ams), London (lon), and finally, Los Angeles (lax). Here, the traffic transfers to cenic.net, which delivers the packets to the caida.org host within the network of the San Diego Supercomputer Center (sdsc) in La Jolla, CA.

At hop 8, we see a star in place of one of the round trip times. This notation means that no response (error packet) was received in response to the probe. In this case, the cause is probably congestion, but that is not the only possibility. `traceroute` relies on low-priority ICMP packets, which many routers are smart enough to drop in preference to “real” traffic. A few stars shouldn’t send you into a panic.

If you see stars in all the time fields for a given gateway, no “time exceeded” messages came back from that machine. Perhaps the gateway is simply down. Sometimes, a gateway or firewall is configured to silently discard packets with expired TTLs. In this case, you can still see through the silent host to the gateways beyond. Another possibility is that the gateway’s error packets are slow to return and that `traceroute` has stopped waiting for them by the time they arrive.
Some firewalls block ICMP “time exceeded” messages entirely. If such a firewall lies along the path, you won’t get information about any of the gateways beyond it. However, you can still determine the total number of hops to the destination because the probe packets eventually get all the way there.

Also, some firewalls may block the outbound UDP datagrams that traceroute sends to trigger the ICMP responses. This problem causes traceroute to report no useful information at all. If you find that your own firewall is preventing you from running traceroute, make sure the firewall has been configured to pass UDP ports 33434–33534 as well as ICMP ECHO (type 8) packets.

A slow link does not necessarily indicate a malfunction. Some physical networks have a naturally high latency; UMTS/EDGE/GPRS wireless networks are a good example. Sluggishness can also be a sign of high load on the receiving network. Inconsistent round trip times would support such a hypothesis.

You may occasionally see the notation !N instead of a star or round trip time. The notation indicates that the current gateway sent back a “network unreachable” error, meaning that it doesn’t know how to route your packet. Other possibilities include !H for “host unreachable” and !P for “protocol unreachable.” A gateway that returns any of these error messages is usually the last hop you can get to. That host often has a routing problem (possibly caused by a broken network link): either its static routes are wrong or dynamic protocols have failed to propagate a usable route to the destination.

If traceroute doesn’t seem to be working for you or is working slowly, it may be timing out while trying to resolve the hostnames of gateways through DNS. If DNS is broken on the host you are tracing from, use traceroute -n to request numeric output. This option disables hostname lookups; it may be the only way to get traceroute to function on a crippled network.

traceroute needs root privileges to operate. To be available to normal users, it must be installed setuid root. Several Linux distributions include the traceroute command but turn off the setuid bit. Depending on your environment and needs, you can either turn the setuid bit back on or give interested users access to the command through sudo.

Recent years have seen the introduction of several new traceroute-like utilities that can bypass ICMP-blocking firewalls. See the PERTKB Wiki for an overview of these tools at goo.gl/fXpMeu. We especially like mtr, which has a top-like interface and shows a sort of live traceroute. Neat!

When debugging routing issues, look at your site from the perspective of the outside world. Several web-based route tracing services let you do this sort of inverse traceroute right from a browser window. Thomas Kernen maintains a list of these services at traceroute.org.
**Packet sniffers**

**tcpdump** and Wireshark belong to a class of tools known as packet sniffers. They listen to network traffic and record or print packets that meet criteria of your choice. For example, you can inspect all packets sent to or from a particular host, or TCP packets related to one particular network connection.

Packet sniffers are useful both for solving problems that you know about and for discovering entirely new problems. It’s a good idea to take an occasional sniff of your network to make sure the traffic is in order.

Packet sniffers need to be able to intercept traffic that the local machine would not normally receive (or at least, pay attention to), so the underlying network hardware must allow access to every packet. Broadcast technologies such as Ethernet work fine, as do most other modern local area networks.

See [this page](#) for more information about network switches.

Since packet sniffers need to see as much of the raw network traffic as possible, they can be thwarted by network switches, which by design try to limit the propagation of “unnecessary” packets. However, it can still be informative to try out a sniffer on a switched network. You may discover problems related to broadcast or multicast packets. Depending on your switch vendor, you may be surprised at how much traffic you can see. Even if you don’t see other systems’ network traffic, a sniffer can be helpful when you are tracking down problems that involve the local host.

In addition to having access to all network packets, the interface hardware must transport those packets up to the software layer. Packet addresses are normally checked in hardware, and only broadcast/multicast packets and those addressed to the local host are relayed to the kernel. In “promiscuous mode,” an interface lets the kernel read all packets on the network, even the ones intended for other hosts.

Packet sniffers understand many of the packet formats used by standard network services, and they can print these packets in human-readable form. This capability makes it easier to track the flow of a conversation between two programs. Some sniffers print the ASCII contents of a packet in addition to the packet header and so are useful for investigating high-level protocols.

Since some protocols send information (and even passwords) across the network as cleartext, take care not to invade the privacy of your users. On the other hand, nothing quite dramatizes the need for cryptographic security like the sight of a plaintext password captured in a network packet.

Sniffers read data from a raw network device, so they must run as root. Although this root limitation serves to decrease the chance that normal users will listen in on your network traffic, it
is really not much of a barrier. Some sites choose to remove sniffer programs from most hosts to reduce the chance of abuse. If nothing else, you should check your systems' interfaces to be sure they are not running in promiscuous mode without your knowledge or consent.

**tcpdump**: command-line packet sniffer

tcpdump, yet another amazing network tool by Van Jacobson, runs on most systems. tcpdump has long been the industry-standard sniffer, and most other network analysis tools read and write trace files in tcpdump format, also known as libpcap format.

By default, tcpdump tunes in on the first network interface it comes across. If it chooses the wrong interface, you can force an interface with the -i flag. If DNS is broken or you just don’t want tcpdump doing name lookups, use the -n option. This option is important because slow DNS service can cause the filter to start dropping packets before they can be dealt with by tcpdump.

The -v flag increases the information you see about packets, and -vv gives you even more data. Finally, tcpdump can store packets to a file with the -w flag and can read them back in with the -r flag.

Note that tcpdump -w saves only packet headers by default. This default makes for small dumps, but the most helpful and relevant information may be missing. So, unless you are sure you need only headers, use the -s option with a value on the order of 1560 (actual values are MTU-dependent) to capture whole packets for later inspection.

As an example, the following truncated output comes from the machine named nubark. The filter specification host bull limits the display of packets to those that directly involve the machine bull, either as source or as destination.

```
$ sudo tcpdump host bull
12:35:23.519961 nubark.domain > bull.41537: A 66.77.122.161 (112) (DF)
```

The first packet shows bull sending a DNS lookup request about atrust.com to nubark. The response is the IP address of the machine associated with that name, which is 66.77.122.161. Note the time stamp on the left and tcpdump’s understanding of the application-layer protocol (in this case, DNS). The port number on bull is arbitrary and is shown numerically (41537), but since the server port number (53) is well known, tcpdump shows its symbolic name, domain.

Packet sniffers can produce an overwhelming amount of information—overwhelming not only for you but also for the underlying operating system. To avoid this problem on busy networks, tcpdump lets you specify complex filters. For example, the following filter collects only incoming web traffic from one subnet:

```
$ sudo tcpdump src net 192.168.1.0/24 and dst port 80
```
The **tcpdump** man page contains several good examples of advanced filtering along with a complete listing of primitives.

**Wireshark and TShark: tcpdump on steroids**

**tcpdump** has been around since approximately the dawn of time, but a newer open source package called Wireshark (formerly known as Ethereal) has been gaining ground rapidly. Wireshark is under active development and incorporates more functionality than most commercial sniffing products. It’s an incredibly powerful analysis tool and should be included in every networking expert’s tool kit. It’s also an invaluable learning aid.

Wireshark includes both a GUI interface (**wireshark**) and a command-line interface (**tshark**). It’s available as a core package on most operating systems. If it’s not in your system’s core repository, check wireshark.org, which hosts the source code and a variety of precompiled binaries.

Wireshark can read and write trace files in the formats used by many other packet sniffers. Another handy feature is that you can click on any packet in a TCP conversation and ask Wireshark to reassemble (splice together) the payload data of all the packets in the stream. This feature is useful if you want to examine the data transferred during a complete TCP exchange, such as a connection on which an email message is transmitted across the network.

Wireshark’s capture filters are functionally identical to **tcpdump**’s since Wireshark uses the same underlying **libpcap** library. Watch out, though—one important gotcha with Wireshark is the added feature of “display filters,” which affect what you see rather than what’s actually captured by the sniffer. The display filter syntax is more powerful than the **libpcap** syntax supported at capture time. The display filters do look somewhat similar, but they are not the same.

---

See [this page](#) for more information about SANs.

---

Wireshark has built-in dissectors for a wide variety of network protocols, including many used to implement SANs. It breaks packets into a structured tree of information in which every bit of the packet is described in plain English.

A note of caution regarding Wireshark: although it has lots of neat features, it has also required many security updates over the years. Run a current copy, and do not leave it running indefinitely on sensitive machines; it might be a potential route of attack.
13.13 Network Monitoring

Chapter 28, Monitoring, describes several general-purpose platforms that can help structure the ongoing oversight of your systems and networks. These systems accept data from a variety of sources, summarize it in a way that illuminates ongoing trends, and alert administrators to problems that require immediate attention.

The network is a key component of any computing environment, so it’s often one of the first parts of the infrastructure to benefit from systematic monitoring. If you don’t feel quite ready to commit to a single monitoring platform for all your administrative needs, the packages outlined in this section are good options for small-scale monitoring that’s focused on the network.
SmokePing: gather ping statistics over time

Even healthy networks drop an occasional packet. On the other hand, networks should not drop packets regularly, even at a low rate, because the impact on users can be disproportionately severe. Because high-level protocols often function even in the presence of packet loss, you might never notice dropped packets unless you’re actively monitoring for them.

SmokePing, an open source tool by Tobias Oetiker, can help you develop a more comprehensive picture of your networks’ behavior. SmokePing sends several ping packets to a target host at regular intervals. It shows the history of each monitored link through a web front end and can send alarms when things go amiss. You can get a copy from oss.oetiker.ch/smokeping.

Exhibit D shows a SmokePing graph. The vertical axis is the round trip time of pings, and the horizontal axis is time (weeks). The black line from which the gray spikes stick up indicates the median round trip time. The spikes themselves are the transit times of individual packets. Since the gray in this graph appears only above the median line, the great majority of packets must be traveling at close to the median speed, with just a few being delayed. This is a typical finding.

Exhibit D: Sample SmokePing graph

The stair-stepped shape of the median line indicates that the baseline transit time to this destination has changed several times during the monitoring period. The most likely hypotheses to explain this observation are either that the host is reachable by several routes or that it is actually a collection of several hosts that have the same DNS name but multiple IP addresses.
iPerf: track network performance

Ping-based tools are helpful for verifying reachability, but they’re not really powerful enough to analyze and track network performance. Enter iPerf. The latest version, iPerf3, has an extensive set of features that administrators can use to fine tune network settings for maximum performance.

Here, we look only at iPerf’s throughput monitoring. At the most basic level, iPerf opens a connection (TCP or UDP) between two servers, passes data between them, and records how long the process took.

Once you’ve installed **iperf** on both machines, start the server side.

```
$ iperf -s

Server listening on TCP port 5001
TCP window size: 85.3 KByte (default)
```

Then, on the machine you want to test from, transfer some data as shown here.

```
$ iperf -c 10.211.55.11

Client connecting to 10.211.55.11, TCP port 5001
TCP window size: 22.5 KByte (default)

[  3] local 10.211.55.10 port 53862 connected with 10.211.55.11 port 5001
[ ID] Interval      Transfer     Bandwidth
[  3] 0.0-10.0 sec  4.13 GBytes  3.55 Gbits/sec
```

iPerf returns great instantaneous data for tracking bandwidth. It’s particularly helpful for assessing the effect of changes to kernel parameters that control the network stack, such as changes to the maximum transfer unit (MTU); see [this page](#) for more details.
Cacti: collect and graph data

Cacti, available from cacti.net, offers several attractive features. It uses a separate package, RRDtool, as its back end, in which it stores monitoring data in the form of zero-maintenance, statically sized databases.

Cacti stores only enough data to create the graphs you want. For example, Cacti could store one sample every minute for a day, one sample every hour for a week, and one sample every week for a year. This consolidation scheme lets you maintain important historical context without having to store unimportant details or spend time on database administration.

Cacti can record and graph any SNMP variable (see this page), as well as many other performance metrics. You’re free to collect whatever data you want. When combined with the NET-SNMP agent, Cacti generates a historical perspective on almost any system or network resource.

Exhibit E shows some examples of graphs created by Cacti. These graphs show the load average on a device over a period of multiple weeks along with a day’s traffic on a network interface.

Exhibit E: Examples of Cacti graphs

Cacti sports easy web-based configuration as well as all the other built-in benefits of RRDtool, such as low maintenance and beautiful graphing. See the RRDtool home page at rrdtool.org for
links to the current versions of RRDtool and Cacti, as well as dozens of other monitoring tools.
13.14 **Firewalls and NAT**

*We do not* recommend the use of Linux, UNIX, or Windows systems as firewalls because of the insecurity inherent in running a full-fledged, general-purpose operating system. However, all operating systems have firewall features, and a hardened system is a workable substitute for organizations that don’t have the budget for a high-dollar firewall appliance. Likewise, a Linux or UNIX firewall is a fine option for a security-savvy home user with a penchant for tinkering. (That said, many consumer-oriented networking devices, such as Linksys’s router products, use Linux and `iptables` at their core.)

If you are set on using a general-purpose computer as a firewall, make sure that it’s up to date with respect to security configuration and patches. A firewall machine is an excellent place to put into practice all the recommendations found in Chapter 27, *Security*. (The section that starts [here](#) discusses packet-filtering firewalls in general. If you are not familiar with the basic concept of a firewall, it would probably be wise to read that section before continuing.)

Microsoft has largely succeeded in convincing the world that every computer needs its own built-in firewall. However, that’s not really true. In fact, machine-specific firewalls can lead to no end of inconsistent behavior and mysterious network problems if they are not managed in synchrony with site-wide standards.

Two main schools of thought deal with the issue of machine-specific firewalls. The first school considers them superfluous. According to this view, firewalls belong on gateway routers, where they can protect an entire network through the application of one consistent (and consistently applied) set of rules.

The second school considers machine-specific firewalls an important component of a “defense in depth” security plan. Although gateway firewalls are theoretically sufficient to control network traffic, they can be compromised, routed around, or administratively misconfigured. Therefore, it’s prudent to implement the same network traffic restrictions through multiple, redundant firewall systems.

If you do choose to implement machine-specific firewalls, you need a system for deploying them in a consistent and easily updatable way. The configuration management systems described in Chapter 23 are excellent candidates for this task. Don’t rely on manual configuration; it’s just too vulnerable to entropy.
Version 2.4 of the Linux kernel introduced an all-new packet-handling engine, called Netfilter, along with a command-line tool, **iptables**, to manage it. An even newer system, nftables, has been available since kernel version 3.13 from 2014. It’s an elaboration of the Netfilter system that’s configured with the **nft** command rather than the **iptables** command. We don’t discuss nftables in this book, but it’s worth evaluating at sites that run current kernels.

**iptables** configuration can be rather fiddly. Debian and Ubuntu include a simple front end, **ufw**, that facilitates common operations and configurations. It’s worth checking out if your needs don’t stray far from the mainstream.

**iptables** applies ordered “chains” of rules to network packets. Sets of chains make up “tables” and are used for handling specific kinds of traffic.

For example, the default **iptables** table is named “filter”. Chains of rules in this table are used for packet-filtering network traffic. The filter table contains three default chains: FORWARD, INPUT, and OUTPUT. Each packet handled by the kernel is passed through exactly one of these chains.

Rules in the FORWARD chain are applied to all packets that arrive on one network interface and need to be forwarded to another. Rules in the INPUT and OUTPUT chains are applied to traffic addressed to or originating from the local host, respectively. These three standard chains are usually all you need for firewalling between two network interfaces. If necessary, you can define a custom configuration to support more complex accounting or routing scenarios.

In addition to the filter table, **iptables** includes the “nat” and “mangle” tables. The nat table contains chains of rules that control Network Address Translation (here, “nat” is the name of the **iptables** table and “NAT” is the name of the generic address translation scheme). The section Private addresses and network address translation (NAT) discusses NAT, and an example of the nat table in action is shown here. Later in this section, we use the nat table’s PREROUTING chain for antispooﬁng packet ﬁltering.

The mangle table contains chains that modify or alter the contents of network packets outside the context of NAT and packet ﬁltering. Although the mangle table is handy for special packet handling, such as resetting IP time-to-live values, it is not typically used in most production environments. We discuss only the filter and nat tables in this section, leaving the mangle table to the adventurous.

**iptables** rule targets

Each rule that makes up a chain has a “target” clause that determines what to do with matching packets. When a packet matches a rule, its fate is in most cases sealed; no additional rules are checked. Although many targets are defined internally to **iptables**, it is possible to specify another chain as a rule’s target.
The targets available to rules in the filter table are ACCEPT, DROP, REJECT, LOG, ULOG, REDIRECT, RETURN, MIRROR, and QUEUE. When a rule results in an ACCEPT, matching packets are allowed to proceed on their way. DROP and REJECT both drop their packets; DROP is silent, and REJECT returns an ICMP error message. LOG gives you a simple way to track packets as they match rules, and ULOG expands logging.

REDIRECT shunts packets to a proxy instead of letting them go on their merry way. For example, you might use this feature to force all your site’s web traffic to go through a web cache such as Squid. RETURN terminates user-defined chains and is analogous to the return statement in a subroutine call. The MIRROR target swaps the IP source and destination addresses before sending the packet. Finally, QUEUE hands packets to local user programs through a kernel module.

**iptables firewall setup**

Before you can use **iptables** as a firewall, you must enable IP forwarding and make sure that various **iptables** modules have been loaded into the kernel. For more information on enabling IP forwarding, see [Linux TCP/IP options](#) or [Security-related kernel variables](#). Packages that install **iptables** generally include startup scripts to achieve this enabling and loading.

A Linux firewall is usually implemented as a series of **iptables** commands contained in an **rc** startup script. Individual **iptables** commands usually take one of the following forms:

```
iptables -F chain-name
iptables -P chain-name target
iptables -A chain-name -i interface -j target
```

The first form (-F) flushes all prior rules from the chain. The second form (-P) sets a default policy (aka target) for the chain. We recommend that you use DROP for the default chain target. The third form (-A) appends the current specification to the chain. Unless you specify a table with the -t argument, your commands apply to chains in the filter table. The -i parameter applies the rule to the named interface, and -j identifies the target. **iptables** accepts many other clauses, some of which are shown in Table 13.10.

**Table 13.10: Command-line flags for iptables filters**
A complete example

Below, we break apart a complete example. We assume that the eth1 interface goes to the Internet and that the eth0 interface goes to an internal network. The eth1 IP address is 128.138.101.4, the eth0 IP address is 10.1.1.1, and both interfaces have a netmask of 255.255.255.0. This example uses stateless packet filtering to protect the web server with IP address 10.1.1.2, which is the standard method of protecting Internet servers. Later in the example, we show how to use stateful filtering to protect desktop users.

Our first set of rules initializes the filter table. First, all chains in the table are flushed, then the INPUT and FORWARD chains’ default target is set to DROP. As with any other network firewall, the most secure strategy is to drop any packets you have not explicitly allowed.

```
iptables -F
iptables -P INPUT DROP
iptables -P FORWARD DROP
```

Since rules are evaluated in order, we put our busiest rules at the front. However, we’re careful to ensure that reordering the rules for performance doesn’t modify functionality.

The first rule allows all connections through the firewall that originate from within the trusted net. The next three rules in the FORWARD chain allow connections through the firewall to network services on 10.1.1.2. Specifically, we allow SSH (port 22), HTTP (port 80), and HTTPS (port 443) through to our web server.

```
iptables -A FORWARD -i eth0 -p ANY -j ACCEPT
iptables -A FORWARD -d 10.1.1.2 -p tcp --dport 22 -j ACCEPT
iptables -A FORWARD -d 10.1.1.2 -p tcp --dport 80 -j ACCEPT
iptables -A FORWARD -d 10.1.1.2 -p tcp --dport 443 -j ACCEPT
```

The only TCP traffic we allow to our firewall host (10.1.1.1) is SSH, which is useful for managing the firewall itself. The second rule listed below allows loopback traffic, which stays local to the host. Administrators get nervous when they can’t ping their default route, so the third rule here allows ICMP ECHO_REQUEST packets from internal IP addresses.
For any IP host to work properly on the Internet, certain types of ICMP packets must be allowed through the firewall. The following eight rules allow a minimal set of ICMP packets to the firewall host, as well as to the network behind it.

```
iptables -A INPUT -i eth0 -d 10.1.1.1 -p tcp --dport 22 -j ACCEPT
iptables -A INPUT -i lo -d 127.0.0.1 -p ANY -j ACCEPT
iptables -A INPUT -i eth0 -d 10.1.1.1 -p icmp --icmp-type 8 -j ACCEPT
```

See this page for more information about IP spoofing.

We next add rules to the PREROUTING chain in the nat table. Although the nat table is not intended for packet filtering, its PREROUTING chain is particularly useful for antispoofing filtering. If we put DROP entries in the PREROUTING chain, they need not be present in the INPUT and FORWARD chains, since the PREROUTING chain is applied to all packets that enter the firewall host. It’s cleaner to put the entries in a single place rather than to duplicate them.

```
iptables -t nat -A PREROUTING -i eth1 -s 10.0.0.0/8 -j DROP
iptables -t nat -A PREROUTING -i eth1 -s 172.16.0.0/12 -j DROP
iptables -t nat -A PREROUTING -i eth1 -s 192.168.0.0/16 -j DROP
iptables -t nat -A PREROUTING -i eth1 -s 127.0.0.0/8 -j DROP
iptables -t nat -A PREROUTING -i eth1 -s 224.0.0.0/4 -j DROP
```

Finally, we end both the INPUT and FORWARD chains with a rule that forbids all packets not explicitly permitted. Although we already enforced this behavior with the `iptables -P` commands, the LOG target lets us see who is knocking on our door from the Internet.

```
iptables -A INPUT -i eth1 -j LOG
iptables -A FORWARD -i eth1 -j LOG
```

Optionally, we could set up IP NAT to disguise the private address space used on the internal network. See this page for more information about NAT.

One of the most powerful features that Netfilter brings to Linux firewalls is stateful packet filtering. Instead of allowing specific incoming services, a firewall for clients connecting to the Internet needs to allow incoming responses to the client’s requests. The simple stateful
FORWARD chain below allows all traffic to leave our network but allows only incoming traffic that’s related to connections initiated by our hosts.

```
iptables -A FORWARD -i eth0 -p ANY -j ACCEPT
iptables -A FORWARD -m state --state ESTABLISHED,RELATED -j ACCEPT
```

Certain kernel modules must be loaded to enable `iptables` to track complex network sessions such as those of FTP and IRC. If these modules are not loaded, `iptables` simply disallows those connections. Although stateful packet filters can increase the security of your site, they also add to the complexity of the network and can reduce performance. Be sure you need stateful functionality before implementing it in your firewall.

Perhaps the best way to debug your `iptables` rulesets is to use `iptables -L -v`. These options tell you how many times each rule in your chains has matched a packet. We often add temporary `iptables` rules with the LOG target when we want more information about the packets that get matched. You can often solve trickier problems by using a packet sniffer such as `tcpdump`.

**Linux NAT and packet filtering**

Linux traditionally implements only a limited form of Network Address Translation (NAT) that is more properly called Port Address Translation, or PAT. Instead of using a range of IP addresses as a true NAT implementation would, PAT multiplexes all connections onto a single address. The details and differences aren’t of much practical importance.

`iptables` implements NAT as well as packet filtering. In earlier versions of Linux this functionality was a bit of a mess, but `iptables` makes a much cleaner separation between the NAT and filtering features. Of course, if you use NAT to let local hosts access the Internet, you must use a full complement of firewall filters as well.

To make NAT work, enable IP forwarding in the kernel by setting the kernel variable `/proc/sys/net/ipv4/ip_forward` to 1. Additionally, insert the appropriate kernel modules:

```
$ sudo modprobe iptable_nat
$ sudo modprobe ip_conntrack
$ sudo modprobe ip_conntrack_ftp
```

Many other modules track connections; see the `net/netfilter` subdirectory underneath `/lib/modules` for a more complete list and enable the ones you need.

The `iptables` command to route packets using NAT is of the form

```
iptables -t nat -A POSTROUTING -o eth0 -j SNAT --to 128.138.101.4
```

This example is for the same host as the filtering example in the previous section, so eth1 is the interface connected to the Internet. The eth1 interface does not appear directly in the command line above, but its IP address is the one that appears as the argument to `--to`. The eth0 interface is the one connected to the internal network.
To Internet hosts, it appears that all packets from hosts on the internal network have eth1’s IP address. The host that implements NAT receives incoming packets, looks up their true destinations, rewrites them with the appropriate internal network IP address, and sends them on their merry way.
IPFilter for UNIX systems

IPFilter, an open source package developed by Darren Reed, supplies NAT and stateful firewall services on a variety of systems, including Linux and FreeBSD. You can use IPFilter as a loadable kernel module (which is recommended by the developers) or include it statically in the kernel.

IPFilter is mature and feature-complete. The package has an active user community and a history of continuous development. It is capable of stateful tracking even for stateless protocols such as UDP and ICMP.

IPFilter reads filtering rules from a configuration file (usually `/etc/ipf/ipf.conf` or `/etc/ipf.conf`) rather than obliging you to run a series of commands as does `iptables`. An example of a simple rule that could appear in `ipf.conf` is

```
block in all
```

This rule blocks all inbound traffic (i.e., network activity received by the system) on all network interfaces. Certainly secure, but not particularly useful!

Table 13.11 shows some of the possible conditions that can appear in an `ipf` rule.

<table>
<thead>
<tr>
<th>Condition</th>
<th>Meaning or possible values</th>
</tr>
</thead>
<tbody>
<tr>
<td><code>on interface</code></td>
<td>Applies the rule to the specified interface</td>
</tr>
<tr>
<td><code>proto protocol</code></td>
<td>Selects packet according to protocol: tcp, udp, or icmp</td>
</tr>
<tr>
<td><code>from source-ip</code></td>
<td>Filters by source: host, network, or any</td>
</tr>
<tr>
<td><code>to dest-ip</code></td>
<td>Filters by destination: host, network, or any</td>
</tr>
<tr>
<td><code>port = port#</code></td>
<td>Filters by port name (from <code>/etc/services</code>) or number</td>
</tr>
<tr>
<td><code>flags flag-spec</code></td>
<td>Filters according to TCP header flags bits</td>
</tr>
<tr>
<td><code>icmp-type number</code></td>
<td>Filters by ICMP type and code</td>
</tr>
<tr>
<td><code>keep state</code></td>
<td>Retains details about the flow of a session; see below</td>
</tr>
</tbody>
</table>

| a. You can use any comparison operator: =, <, >, <=, >=, etc. |

IPFilter evaluates rules in the sequence in which they are presented in the configuration file. The `last` match is binding. For example, inbound packets traversing the following filter will always pass:

```
block in all
pass in all
```

The `block` rule matches all packets, but so does the `pass` rule, and `pass` is the last match. To force a matching rule to apply immediately and make IPFilter skip subsequent rules, use the `quick`
An industrial-strength firewall typically contains many rules, so liberal use of `quick` is important in order to maintain the performance of the firewall. Without it, every packet is evaluated against every rule, and this wastefulness is costly.

Perhaps the most common use of a firewall is to control access to and from a specific network or host, often with respect to a specific port. IPFilter has powerful syntax to control traffic at this level of granularity. In the following rules, inbound traffic is permitted to the 10.0.0.0/24 network on TCP ports 80 and 443 and on UDP port 53.

```
block out quick all
pass in quick proto tcp from any to 10.0.0.0/24 port = 80 keep state
pass in quick proto tcp from any to 10.0.0.0/24 port = 443 keep state
pass in quick proto udp from any to 10.0.0.0/24 port = 53 keep state
block in all
```

The `keep state` keywords deserve special attention. IPFilter can keep track of connections by noting the first packet of new sessions. For example, when a new packet arrives addressed to port 80 on 10.0.0.10, IPFilter makes an entry in the state table and allows the packet through. It also allows the reply from the web server even though the first rule explicitly blocks all outbound traffic.

`keep state` is also useful for devices that offer no services but that must initiate connections. The following ruleset permits all conversations that are initiated by 192.168.10.10. It blocks all inbound packets except those related to connections that have already been initiated.

```
block in quick all
pass out quick from 192.168.10.10/32 to any keep state
```

The `keep state` keywords work for UDP and ICMP packets, too, but since these protocols are stateless, the mechanics are slightly more ad hoc: IPFilter permits responses to a UDP or an ICMP packet for 60 seconds after the inbound packet is seen by the filter. For example, if a UDP packet from 10.0.0.10, port 32,000, is addressed to 192.168.10.10, port 53, a UDP reply from 192.168.10.10 will be permitted until 60 seconds have passed. Similarly, an ICMP echo reply (ping response) is permitted after an echo request has been entered in the state table.

See this page for more information about NAT.

IPFilter uses the `map` keyword (in place of `pass` and `block`) to provide NAT services. In the following rule, traffic from the 10.0.0.0/24 network is mapped to the current routable address on the em0 interface.
map em0 10.0.0.0/24 -> 0/32

The filter must be reloaded if the address of em0 changes, as might happen if em0 leases a dynamic IP address through DHCP. For this reason, IPFilter’s NAT features are best used at sites that have a static IP address on the Internet-facing interface.

Table 13.12 lists the command-line tools that come with the IPFilter package.

<table>
<thead>
<tr>
<th>Cmd</th>
<th>Function</th>
</tr>
</thead>
<tbody>
<tr>
<td>ipf</td>
<td>Manages rules and filter lists</td>
</tr>
<tr>
<td>ipfstat</td>
<td>Obtains statistics about packet filtering</td>
</tr>
<tr>
<td>ipmon</td>
<td>Monitors logged filter information</td>
</tr>
<tr>
<td>ipnat</td>
<td>Manages NAT rules</td>
</tr>
</tbody>
</table>

Of the commands in Table 13.12, ipf is the most commonly used. ipf accepts a rule file as input and adds correctly parsed rules to the kernel’s filter list. ipf adds rules to the end of the filter unless you use the -Fa argument, which flushes all existing rules. For example, to flush the kernel’s existing set of filters and load the rules from ipf.conf, use the following syntax:

```
$ sudo ipf -Fa -f /etc/ipf/ipf.conf
```

IPFilter relies on pseudo-device files in /dev for access control, and by default only root can edit the filter list. We recommend leaving the default permissions in place and using sudo to maintain the filter.

Use ipf’s -v flag when loading the rules file to debug syntax errors and other problems in the configuration.
13.15 **Cloud networking**

One of the interesting features of the cloud is that you get to define the networking environment in which your virtual servers live. Ultimately, of course, cloud servers live on physical computers that are connected to real network hardware. However, that doesn’t necessarily mean that virtual servers running on the same node are networked together. The combination of virtualization technology and programmable network switching equipment gives platform providers great flexibility to define the networking model they export to clients.
AWS’s virtual private cloud (VPC)

VPC, the software-defined network technology for Amazon Web Services, creates private networks within the broader AWS network. VPC was first introduced in 2009 as a bridge between an on-premises data center and the cloud, opening up many hybrid use cases for enterprise organizations. Today, VPC is a central feature of AWS, and a default VPC is included for all accounts. EC2 instances for newer AWS accounts must be created within a VPC, and most new AWS services launch with native VPC support. (Longtime users gripe that AWS services are incomplete until they support VPC.)

The central features of VPC include

- An IPv4 address range selected from the RFC1918 private address space, expressed in CIDR notation (for example, 10.110.0.0/16 for the addresses 10.110.0.0-10.110.255.255; VPC has also recently added support for IPv6)
- Subnets to segment the VPC address space into smaller subnetworks
- Routing tables that determine where to send traffic
- Security groups that act as firewalls for EC2 instances
- Network Access Control Lists (NACLs) to isolate subnets from each other

You can create as many VPCs as you need, and no other AWS customer has access to network traffic within your VPCs. Depending on the state of your account, AWS may initially limit you to 5 VPCs. However, you can request a higher limit if you need it.

VPCs within the same region can be peered, creating private routes between separate networks. VPCs in different regions can be connected with software VPN tunnels over the Internet, or with expensive, custom, direct connections to AWS data centers over private circuits that you must lease from a telco.

VPCs can be as small as a /28 network or as large as a /16. It’s important to plan ahead because the size cannot be adjusted after the VPC is created. Choose an address space that is large enough to accommodate future growth, but also ensure that it does not conflict with other networks that you may wish to connect.

Subnets and routing tables

Like traditional networks, VPCs are divided into subnets. Public subnets are for servers that must talk directly to clients on the Internet. They are akin to traditional DMZs. Private subnets are inaccessible from the Internet and are intended for trusted or sensitive systems.

See this page for more information about DMZs.
VPC routing is simpler than routing for a traditional hardware network because the cloud does not simulate physical topology. Every accessible destination is reachable in one logical hop.

In the world of physical networking, every device has a routing table that tells it how to route outbound network packets. But in VPC, routing tables are also an abstract entity that’s defined through the AWS web console or its command-line equivalent. Every VPC subnet has an associated VPC routing table. When instances are created on a subnet, their routing tables are initialized from the VPC template.

The simplest routing table contains only a default static route for reaching other instances within the same VPC. You can add additional routes to access the Internet, on-premises networks (through VPN connections), or other VPCs (through peering connections).

A component called an Internet Gateway connects VPCs to the Internet. This entity is transparent to the administrator and is managed by AWS. However, you need to create one and attach it to your VPC if instances are to have Internet connectivity. Hosts in public subnets can access the Internet Gateway directly.

Instances in private subnets cannot be reached from the Internet even if they are assigned public IP addresses, a fact that results in much confusion for new users. For outbound access, they must hop through a NAT gateway on a public subnet. VPC offers a managed NAT feature which saves you the overhead of running your own gateway, but it incurs an additional hourly cost. The NAT gateway is a potential bottleneck for applications that have high throughput requirements, so it’s better to locate the servers for such applications on public subnets, avoiding NAT.

AWS’s implementation of IPv6 does not have NAT, and all instances set up for IPv6 receive “public” (i.e., routable) IPv6 addresses. You make IPv6 subnets private by connecting them through an egress-only Internet Gateway (aka eigw) which blocks inbound connections. The gateway is stateful, so external hosts can talk to servers on the private IPv6 network as long as the AWS server initiates the connection.

To understand the network routing for an instance, you’ll find it more informative to review the VPC routing table for its subnet than to look at the instance’s actual routing table (such as might be displayed by `netstat -r` or `ip route show` when logged in to the instance). The VPC version identifies gateways (“targets”) by their AWS identifiers, which makes the table easy to parse at a glance.

In particular, you can easily distinguish public subnets from private subnets by looking at the VPC routing table. If the default gateway (i.e., the target associated with the address 0.0.0.0/0) is an Internet Gateway (an entity named igw-something), then that subnet is public. If the default gateway is a NAT device (a route target prefixed by an instance ID, i-something, or nat-something), then the subnet is private.

*Table 13.13* shows an example routing table for a private subnet.
Table 13.13: Example VPC routing table for a private subnet

<table>
<thead>
<tr>
<th>Destination</th>
<th>Target</th>
<th>Target type</th>
</tr>
</thead>
<tbody>
<tr>
<td>10.110.0.0/16</td>
<td>local</td>
<td>Built-in route for the local VPC network</td>
</tr>
<tr>
<td>0.0.0.0/0</td>
<td>nat-a31ed812</td>
<td>Internet access through a VPC NAT gateway</td>
</tr>
<tr>
<td>10.120.0.0/16</td>
<td>pcx-38c3e8b2</td>
<td>Peering connection to another VPC</td>
</tr>
<tr>
<td>192.168.0.0/16</td>
<td>vgw-1e513d90</td>
<td>VPN gateway to an external network</td>
</tr>
</tbody>
</table>

VPCs are regional, but subnets are restricted to a single availability zone. To build highly available systems, create at least one subnet per zone and distribute instances evenly among all the subnets. A typical design puts load balancers or other proxies in public subnets and restricts web, application, and database servers to private subnets.

Security groups and NACLs

Security groups are firewalls for EC2 instances. Security group rules dictate which source addresses are allowed for ICMP, UDP, and TCP traffic (ingress rules), and which ports on other systems can be accessed by instances (egress rules). Security groups deny all connections by default, so any rules you add allow additional traffic.

All EC2 instances belong to at least one security group, but they may be in as many as five. To be perfectly accurate, security groups are associated with network interfaces, and an instance can have more than one network interface. So we should really say that the maximum number of security groups is the number of network interfaces times five.

The more security groups an instance belongs to, the more confusing it can be to determine precisely what traffic is and is not allowed. We prefer that each instance be in only one security group, even if that configuration results in some duplicate rules among groups.

When adding rules to security groups, always consider the principle of least privilege. Opening ports unnecessarily presents a security risk, especially for systems that have public, routable IP addresses. For example, a web server may only need ports 22 (SSH, used for management and control of the system), 80 (HTTP), and 443 (HTTPS).

In addition, all hosts should accept the ICMP packets used to implement path MTU discovery. Failure to admit these packets can lower network bandwidth considerably, so we find puzzling AWS’s decision to block them by default. See goo.gl/WrETNq (deep link into docs.aws.amazon.com) for the steps to enable these packets.

Most security groups have granular inbound rules but allow all outbound traffic, as shown in Table 13.14. This configuration is convenient since you don’t need to think about what outside connectivity your systems have. However, it’s easier for attackers to set up shop if they can retrieve tools and communicate with their external control systems. The most secure networks have both inbound and outbound restrictions.
Table 13.14: Typical security group rules

<table>
<thead>
<tr>
<th>Direction</th>
<th>Proto</th>
<th>Ports</th>
<th>CIDR</th>
<th>Notes</th>
</tr>
</thead>
<tbody>
<tr>
<td>Ingress</td>
<td>TCP</td>
<td>22</td>
<td>10.110.0.0/16</td>
<td>SSH from the internal network</td>
</tr>
<tr>
<td>Ingress</td>
<td>TCP</td>
<td>80</td>
<td>0.0.0.0/0</td>
<td>HTTP from anywhere</td>
</tr>
<tr>
<td>Ingress</td>
<td>TCP</td>
<td>443</td>
<td>0.0.0.0/0</td>
<td>HTTPS from anywhere</td>
</tr>
<tr>
<td>Ingress</td>
<td>ICMP</td>
<td>n/a</td>
<td>0.0.0.0/0</td>
<td>Allow path MTU discovery</td>
</tr>
<tr>
<td>Egress</td>
<td>ALL</td>
<td>ALL</td>
<td>0.0.0.0/0</td>
<td>Outbound traffic (all OK)</td>
</tr>
</tbody>
</table>

a. See goo.gl/WrETNg for detailed instructions; this record is a bit tricky to set up.

Much like access control lists on a firewall device, NACLs control traffic among subnets. Unlike security groups, NACLs are stateless: they don’t distinguish between new and existing connections. They are similar in concept to NACLs on a hardware firewall. NACLs allow all traffic by default. In the wild, we see security groups used far more often than NACLs.

A sample VPC architecture

Exhibit F depicts two VPCs, each with public and private subnets. Network 2 hosts an Elastic Load Balancer in its public subnets. The ELB acts as a proxy for some autoscaling EC2 instances that live in the private subnet and protects those instances from the Internet. Service 2 in Network 2 may need access to Service 1 hosted in Network 1, and they can communicate privately through VPC peering.

Exhibit F: Peered VPCs with public and private subnets

Architecture diagrams like Exhibit F communicate dense technical details more clearly than written prose. We maintain diagrams like this one for every application we deploy.

Creating a VPC with Terraform
VPCs are composed of many resources, each of which has its own settings and options. The interdependencies among these objects are complex. It’s possible to create and manage almost everything by using the CLI or web console, but that approach requires that you keep all the minutiae in your head. Even if you can keep all the moving parts straight during the initial setup, it’s difficult to track your work over time.

Terraform, a tool from HashiCorp, creates and manages cloud resources. For example, Terraform can create a VPC, launch instances, and then initialize those instances by running scripts or other configuration management tools. Terraform configuration is expressed in HashiCorp Configuration Language (HCL), a declarative format that looks similar to JSON but adds variable interpolation and comments. The file can be tracked in revision control, so it’s simple to update and adapt.

The example below shows a Terraform configuration for a simple VPC with one public subnet. We think it’s rather self-documenting, intelligible even to a neophyte:
The Terraform documentation is the authoritative syntax reference. You’ll find many example configurations like this one in the Terraform GitHub repository and elsewhere on the Internet.

Run **terraform apply** to have Terraform create this VPC. It examines the current directory (by default) for `.tf` files and processes each of them, assembling an execution plan and then invoking API calls in the appropriate order. You can set the AWS API credentials in the configuration file or through the `AWS_ACCESS_KEY_ID` and `AWS_SECRET_ACCESS_KEY` environment variables, as we have done here.
time measures how long it takes to create all the resources in the configuration (about 4.5 seconds). The <computed> values indicate that Terraform chose defaults because we didn’t specify those settings explicitly.

The state of all resources created by Terraform is saved in a file called terraform.tfstate. This file must be preserved so that Terraform knows which resources are under its control. In the future, Terraform will discover the managed resources on its own.

We can clean up the VPC just as easily:
Terraform is cloud-agnostic, so it can manage resources for AWS, GCP, DigitalOcean, Azure, Docker, and other providers.

How do you know when to use Terraform and when to use the CLI? If you’re building infrastructure for a team or project, or if you’ll need to make changes and repeat the build later, use Terraform. If you need to fire off a quick instance as a test, if you need to inspect the details of a resource, or if you need to access the API from a shell script, use the CLI.
**Google Cloud Platform networking**

On the Google Cloud Platform, networking is functionally part of the platform, as opposed to being represented as a distinct service. GCP private networks are global: an instance in the us-east1 region can communicate with another instance in europe-west1 over the private network, a fact that makes it easy to build global network services. Network traffic among instances in the same zone is free, but there is a fee for traffic between zones or regions.

New projects have a default network with the 10.240.0.0/16 address range. You can create up to five separate networks per project, and instances are members of exactly one network. Many sites use this network architecture to isolate test and development from production systems.

Networks can be subdivided by region with subnetworks, a relatively recent addition to GCP that functions differently from subnets on AWS. The global network does not need to be part of a single IPv4 prefix range, and there can be multiple prefixes per region. GCP configures all the routing for you, so instances on different CIDR blocks within the same network can still reach each other. **Exhibit G** demonstrates this topology.

**Exhibit G: A multiregion private GCP network with subnetworks**

There is no concept of a subnetwork being public or private; instead, instances that don’t need to accept inbound traffic from the Internet can simply not have a public, Internet-facing address. Google offers static external IP addresses that you can borrow for use in DNS records without fear that they will be assigned to another customer. When an instance does have an external address, you still won’t see it if you run `ip addr show`; Google handles the address translation for you.
By default, firewall rules in a GCP network apply to all instances. To restrict rules to a smaller set of instances, you can tag instances and filter the rules according to the tags. The default, global firewall rules deny everything except the following:

- ICMP traffic for 0/0
- RDP (remote desktop for Windows, TCP port 3389) for 0/0
- SSH (TCP port 22) for 0/0
- All ports and protocols for the internal network (10.240.0.0/16 by default)

When it comes to decisions that impact security, we always come back to the principle of least privilege. In this case, we recommend narrowing these default rules to block RDP entirely, allow SSH only from your own source IPs, and further restrict traffic within the GCP network. You might also want to block ICMP, but be aware that you need to allow ICMP packets of type 3, code 4 to enable path MTU discovery.
DigitalOcean networking

DigitalOcean does not have a private network, or at least, not one similar to those of GCP and AWS. Droplets can have private interfaces that communicate over an internal network within the same region. However, that network is shared with all other DigitalOcean customers in the same region. This is a slight improvement over using the Internet, but firewalls and in-transit encryption become hard requirements.

We can examine a booted DigitalOcean droplet with the `tugboat` CLI:

```
$ tugboat info ulsah
Droplet fuzzy name provided. Finding droplet ID...done, 8857202
   (ulsah-ubuntu-15-10)
Name:         ulsah-ubuntu-15-10
ID:           8857202
Status:       active
IP4:          45.55.1.165
IP6:          2604:A800:0001:0020:0000:0000:01EF:D001
Private IP:   10.134.131.213
Region:       San Francisco 1 - sf01
Image:        14169855 - ubuntu-15-10-x64
Size:         512MB
Backups Active: false
```

The output includes an IPv6 address in addition to public and private IPv4 addresses.

On the instance, we can further explore by looking at the addresses on the local interfaces.

```
# tugboat ssh ulsah-ubuntu-15-10
# ip address show eth0
2: eth0: <BROADCAST,MULTICAST,UP,LOWER_UP> mtu 1500 qdisc pfifo_fast
   state UP group default qlen 1000
   link/ether 04:01:87:26:d6:01 brd ff:ff:ff:ff:ff:ff
   inet 45.55.1.165/19 brd 45.55.31.255 scope global eth0
       valid_lft forever preferred_lft forever
   inet 10.12.0.8/16 scope global eth0
       valid_lft forever preferred_lft forever
   inet6 fe80::601:87ff:fe26:d601/64 scope link
       valid_lft forever preferred_lft forever
# ip address show eth1
3: eth1: <BROADCAST,MULTICAST,UP,LOWER_UP> mtu 1500 qdisc pfifo_fast
   state UP group default qlen 1000
   link/ether 04:01:87:26:d6:02 brd ff:ff:ff:ff:ff:ff
   inet 10.134.131.213/16 brd 10.134.255.255 scope global eth1
       valid_lft forever preferred_lft forever
   inet6 fe80::601:87ff:fe26:d602/64 scope link
       valid_lft forever preferred_lft forever
```
The public address is assigned directly to the eth0 interface, not translated by the provider as on other cloud platforms. Each interface also has an IPv6 address, so it’s possible to serve traffic through IPv4 and IPv6 simultaneously.
13.16 **Recommended reading**

**History**


Doug Comer’s *Internetworking with TCP/IP* series was for a long time the standard reference for the TCP/IP protocols. The books are designed as undergraduate textbooks and are a good introductory source of background material.

**Salus, Peter H.** *Casting the Net, From ARPANET to INTERNET and Beyond.* Reading, MA: Addison-Wesley Professional, 1995.

This is a lovely history of the ARPANET as it grew into the Internet, written by a historian who has been hanging out with UNIX people long enough to sound like one of them.

An excellent collection of documents about the history of the Internet and its various technologies can be found at [isoc.org/internet/history](http://isoc.org/internet/history).
Classics and bibles


The *UNIX Network Programming* books are the student’s bibles in networking classes that involve programming. If you need only the Berkeley sockets interface, the original edition is still a fine reference. If you need the STREAMS interface too, then the third edition, which includes IPv6, is a good bet. All three are clearly written in typical Rich Stevens style.


*Computer Networks* was the first networking text, and it is still a classic. It contains a thorough description of all the nitty-gritty details going on at the physical and link layers of the protocol stack. The latest edition includes coverage of wireless networks, gigabit Ethernet, peer-to-peer networks, voice over IP, cellular networks, and more.
Protocols


The books in the *TCP/IP Illustrated* series are an excellent and thorough guide to the TCP/IP protocol stack.

Hunt, Craig. *TCP/IP Network Administration (3rd Edition).* Sebastopol, CA: O’Reilly Media, 2002. Like other books in the nutshell series, this book is directed at administrators of UNIX systems. Half the book is about TCP/IP, and the rest deals with higher-level UNIX facilities such as email and remote login.


14 Physical Networking

Regardless of whether your systems live in a data center, a cloud, or an old missile silo, one element they have in common is the need to communicate on a network. The ability to move data quickly and reliably is essential in every environment. If there’s one area in which UNIX technology has touched human lives and influenced other operating systems, it’s in the practical realization of large-scale packetized data transport.

Networks are following the same trail blazed by servers in that the physical and logical views of the network are increasingly separated by a virtualization layer that has its own configuration. That sort of setup is standard in the cloud, but even physical data centers often include a layer of software-defined networking (SDN) these days.

Administrators interact with real-world network hardware less frequently than they once did, but familiarity with traditional networking remains a crucial skill. Virtualized networks closely emulate physical networks in their features, terminology, architecture, and topology.

Many link-layer technologies have been promoted over the years, but Ethernet has emerged as the clear and decisive winner. Now that Ethernet is found on everything from game consoles to refrigerators, a thorough understanding of this technology is critical to success as a system administrator.

Obviously, the speed and reliability of your network have a direct effect on your organization’s productivity. But today, networking has become so pervasive that the state of the network affects even such basic interactions as the ability to make telephone calls. A poorly designed network is
a personal and professional embarrassment that can lead to catastrophic social effects. It can also be expensive to fix.

At least four major factors contribute to success:

- Development of a reasonable network design
- Selection of high-quality hardware
- Proper installation and documentation
- Competent ongoing operations and maintenance

This chapter focuses on understanding, installing, and operating Ethernet networks in an enterprise environment.
14.1 **Ethernet: the Swiss Army Knife of Networking**

Having captured over 95% of the world-wide local area network (LAN) market, Ethernet can be found just about everywhere in its many forms. It started as Bob Metcalfe’s Ph.D. thesis at MIT but is now described in a variety of IEEE standards.

Ethernet was originally specified at 3 Mb/s (megabits per second), but it moved to 10 Mb/s almost immediately. Once a 100 Mb/s standard was finalized in 1994, it became clear that Ethernet would evolve rather than be replaced. This realization touched off a race to build increasingly faster versions of Ethernet, and that race continues today. Table 14.1 highlights the evolution of the various Ethernet standards. We have omitted a few of the less popular Ethernet standards that cropped up along the way.

<table>
<thead>
<tr>
<th>Year</th>
<th>Speed</th>
<th>Common name</th>
<th>IEEE#</th>
<th>Dist</th>
<th>Media*</th>
</tr>
</thead>
<tbody>
<tr>
<td>1973</td>
<td>3 Mb/s</td>
<td>Xerox Ethernet</td>
<td>–</td>
<td>?</td>
<td>Coax</td>
</tr>
<tr>
<td>1976</td>
<td>10 Mb/s</td>
<td>Ethernet 1</td>
<td>–</td>
<td>500m</td>
<td>RG-11 coax</td>
</tr>
<tr>
<td>1989</td>
<td>10 Mb/s</td>
<td>10BASE-T</td>
<td>802.3</td>
<td>100m</td>
<td>Cat 3 UTP copper</td>
</tr>
<tr>
<td>1994</td>
<td>100 Mb/s</td>
<td>100BASE-TX</td>
<td>802.3u</td>
<td>100m</td>
<td>Cat 5 UTP copper</td>
</tr>
<tr>
<td>1999</td>
<td>1 Gb/s</td>
<td>100BASE-T (&quot;gigabit&quot;)</td>
<td>802.3ab</td>
<td>100m</td>
<td>Cat 5e, 6 UTP copper</td>
</tr>
<tr>
<td>2006</td>
<td>10 Gb/s</td>
<td>10GBASE-T (&quot;10 gig&quot;)</td>
<td>802.3an</td>
<td>100m</td>
<td>Cat 6a, 7, 7a UTP</td>
</tr>
<tr>
<td>2009</td>
<td>40 Gb/s</td>
<td>40GBASE-CR4</td>
<td>P802.3ba</td>
<td>10m</td>
<td>UTP copper</td>
</tr>
<tr>
<td></td>
<td></td>
<td>40GBASE-SR4</td>
<td></td>
<td></td>
<td>MM fiber</td>
</tr>
<tr>
<td>2009</td>
<td>100 Gb/s</td>
<td>100GBASE-CR10</td>
<td>P802.3ba</td>
<td>10m</td>
<td>UTP copper</td>
</tr>
<tr>
<td></td>
<td></td>
<td>100GBASE-SR10</td>
<td></td>
<td></td>
<td>MM fiber</td>
</tr>
<tr>
<td>2018b</td>
<td>200 Gb/s</td>
<td>200GBASE-FR4</td>
<td>802.3bs</td>
<td>2km</td>
<td>CWDM fiber</td>
</tr>
<tr>
<td></td>
<td></td>
<td>200GBASE-LR4</td>
<td></td>
<td></td>
<td>CWDM fiber</td>
</tr>
<tr>
<td>2018b</td>
<td>400 Gb/s</td>
<td>400GBASE-SR16</td>
<td>802.3bs</td>
<td>100m</td>
<td>MM fiber (16 strand)</td>
</tr>
<tr>
<td></td>
<td></td>
<td>400GBASE-DR4</td>
<td></td>
<td></td>
<td>MM fiber (4 strand)</td>
</tr>
<tr>
<td></td>
<td></td>
<td>400GBASE-FR8</td>
<td></td>
<td>500m</td>
<td>CWDM fiber</td>
</tr>
<tr>
<td></td>
<td></td>
<td>400GBASE-LR8</td>
<td></td>
<td>10km</td>
<td>CWDM fiber</td>
</tr>
<tr>
<td>2020b</td>
<td>1 Tb/s</td>
<td>TbE</td>
<td>TBD</td>
<td>TBD</td>
<td>TBD</td>
</tr>
</tbody>
</table>

a. MM = Multimode, SM = Single-mode, UTP = Unshielded twisted pair, CWDM = Coarse wavelength division multiplexing  

b. Industry projection  
c. We’ll give the benefit of the doubt and assume this lettering choice was an unfortunate coincidence.
**Ethernet signaling**

The underlying model used by Ethernet can be described as a polite dinner party at which guests (computers) don’t interrupt each other but rather wait for a lull in the conversation (no traffic on the network cable) before speaking. If two guests start to talk at once (a collision) they both stop, excuse themselves, wait a bit, and then one of them starts talking again.

The technical term for this scheme is CSMA/CD:

- **Carrier Sense**: you can tell whether anyone is talking.
- **Multiple Access**: everyone can talk.
- **Collision Detection**: you know when you interrupt someone else.

The actual delay after a collision is somewhat random. This convention avoids the scenario in which two hosts simultaneously transmit to the network, detect the collision, wait the same amount of time, and then start transmitting again, thus flooding the network with collisions. This was not always true!

Today, the importance of the CSMA/CD conventions has been lessened by the advent of switches, which typically limit the number of hosts to two in a given collision domain. (To continue the “dinner party” analogy, you might think of this switched variant of Ethernet as being akin to the scenes found in old movies where two people sit at opposite ends of a long, formal dining table.)
**Ethernet topology**

The Ethernet topology is a branching bus with no loops. A packet can travel between two hosts on the same network in only one way.

Three types of packets can be exchanged on a segment: unicast, multicast, and broadcast. Unicast packets are addressed to only one host. Multicast packets are addressed to a group of hosts. Broadcast packets are delivered to all hosts on a segment.

A “broadcast domain” is the set of hosts that receive packets destined for the hardware broadcast address. Exactly one broadcast domain is defined for each logical Ethernet segment. Under the early Ethernet standards and media (e.g., 10BASE5), physical segments and logical segments were exactly the same because all the packets traveled on one big cable with host interfaces strapped onto the side of it. Attaching a new computer involved boring a hole into the outer sheath of the cable with a special drill to reach the center conductor. A “vampire tap” that bit into the outer conductor was then clamped on with screws.

With the advent of switches, today’s logical segments usually consist of many physical segments (possibly dozens or hundreds) to which only two devices are connected: a switch port and a host. The switches are responsible for escorting multicast and unicast packets to the physical (or wireless) segments on which the intended recipients reside. Broadcast traffic is forwarded to all ports in a logical segment.

A single logical segment can consist of physical (or wireless) segments that operate at different speeds. Hence, switches must have buffering and timing capabilities to let them smooth over any potential timing conflicts.

Wireless networks are another common type of logical Ethernet segment. They behave more like the traditional forms of Ethernet which share one cable among many hosts. See [this page](#).
Unshielded twisted-pair cabling

Unshielded twisted pair (UTP) has historically been the preferred cable medium for Ethernet in most office environments. Today, wireless networking has displaced UTP in many situations. The general “shape” of a UTP network is illustrated in Exhibit A.

Exhibit A: A UTP installation

![UTP installation diagram]

UTP wire is commonly broken down into eight classifications. The performance rating system was first introduced by Anixter, a large cable supplier. These standards were formalized by the Telecommunications Industry Association (TIA) and are known today as Category 1 through Category 8, with a few special variants such as Category 5e and Category 6a thrown in for good measure.

The International Organization for Standardization (ISO) has also jumped into the exciting and highly profitable world of cable classification. They promote standards that are exactly or approximately equivalent to the higher-numbered TIA categories. For example, TIA Category 5 cable is equivalent to ISO Class D cable. For the geeks in the audience, Table 14.2 illustrates the differences among the various modern-day classifications. This is good information to memorize so you can impress your friends at parties.

Table 14.2: UTP cable characteristics
Category 5 cable can support 100 Mb/s and is “table stakes” for network wiring today. Category 5e, Category 6, and Category 6a cabling support 1 Gb/s and are the most common standard currently in use for data cabling. Category 6a is the cable of choice for new installations because it is particularly resistant to interference from older Ethernet signaling standards (e.g., 10BASE-T), a problem that has plagued some Category 5/5e installations. Category 7 and Category 7a cable are intended for 10 Gb/s use, and Category 8 rounds out the family at 40 Gb/s.

Faster standards require multiple pairs of UTP. Having multiple conductors transports data across the link faster than any single pair can support. 100BASE-TX requires two pairs of Category 5 wire. 1000BASE-TX requires four pairs of Category 5e or Category 6/6a wire, and 10GBASE-TX requires four pairs of Category 6a, 7, or 7a wire. All these standards are limited to 100 meters in length.

Both PVC-coated and Teflon-coated wire are available. Your choice of jacketing should depend on the environment in which the cable will be installed. Enclosed areas that feed into the building’s ventilation system (“return air plenums”) typically require Teflon. PVC is less expensive and easier to work with but produces toxic fumes if it catches fire, hence the need to keep it out of air plenums. Check with your fire marshal or local fire department to determine the requirements in your area.

For terminating the four-pair UTP cable at patch panels and RJ-45 wall jacks, we suggest you use the TIA/EIA-568A RJ-45 wiring standard. This standard, which is compatible with other uses of RJ-45 (e.g., RS-232), is a convenient way to keep the wiring at both ends of the connection consistent, regardless of whether you can easily access the cable pairs themselves. Table 14.3 shows the pinouts.

### Table 14.3: TIA/EIA-568A standard for wiring four-pair UTP to an RJ-45 jack

<table>
<thead>
<tr>
<th>Pair</th>
<th>Colors</th>
<th>Wired to</th>
<th>Pair</th>
<th>Colors</th>
<th>Wired to</th>
</tr>
</thead>
<tbody>
<tr>
<td>1</td>
<td>White/Blue</td>
<td>Pins 5/4</td>
<td>3</td>
<td>White/Green</td>
<td>Pins 1/2</td>
</tr>
<tr>
<td>2</td>
<td>White/Orange</td>
<td>Pins 3/6</td>
<td>4</td>
<td>White/Brown</td>
<td>Pins 7/8</td>
</tr>
</tbody>
</table>

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a. NEXT = Near-end crosstalk, ELFEXT = Equal level far-end crosstalk
b. Includes additional TIA and ISO requirements TSB95 and FDAM 2, respectively
Existing building wiring might or might not be suitable for network use, depending on how and when it was installed.
Optical fiber

Optical fiber is used in situations where copper cable is inadequate. Fiber carries signals farther than copper and is also resistant to electrical interference. In cases where fiber isn’t absolutely necessary, copper is normally preferred because it’s less expensive and easier to work with.

“Multimode” and “single mode” fiber are the two common types. Multimode fiber is typically used for applications within a building or campus. It’s thicker than single-mode fiber and can carry multiple rays of light; this feature permits the use of less expensive electronics (e.g., LEDs as a light source).

Single-mode fiber is most often found in long-haul applications, such as intercity or interstate connections. It can carry only a single ray of light and requires expensive precision electronics on the endpoints.

A common strategy to increase the bandwidth across a fiber link is coarse wavelength division multiplexing (CWDM). It’s a way to transmit multiple channels of data through a single fiber on multiple wavelengths (colors) of light. Some of the faster Ethernet standards use this scheme natively. However, it can also be employed to extend the capabilities of an existing dark fiber link through the use of CWDM multiplexers.

TIA-598C recommends color-coding the common types of fiber, as shown in Table 14.4. The key rule to remember is that everything must match. The fiber that connects the endpoints, the fiber cross-connect cables, and the endpoint electronics must all be of the same type and size. Note that although both OM1 and OM2 are colored orange, they are not interchangeable—check the size imprint on the cables to make sure they match. You will experience no end of difficult-to-isolate problems if you don’t follow this rule.

### Table 14.4: Attributes of standard optical fibers

<table>
<thead>
<tr>
<th>Mode</th>
<th>ISO name</th>
<th>Core diameter</th>
<th>Cladding diameter</th>
<th>Color</th>
</tr>
</thead>
<tbody>
<tr>
<td>Multi</td>
<td>OM1</td>
<td>62.5 μm</td>
<td>125 μm</td>
<td>Orange</td>
</tr>
<tr>
<td>Multi</td>
<td>OM2</td>
<td>50 μm</td>
<td>125 μm</td>
<td>Orange</td>
</tr>
<tr>
<td>Multi</td>
<td>OM3</td>
<td>50 μm&lt;sup&gt;b&lt;/sup&gt;</td>
<td>125 μm</td>
<td>Aqua</td>
</tr>
<tr>
<td>Single</td>
<td>OS1</td>
<td>8–10 μm&lt;sup&gt;b&lt;/sup&gt;</td>
<td>125 μm</td>
<td>Yellow</td>
</tr>
</tbody>
</table>

<sup>a</sup> According to ISO 11801

<sup>b</sup> OM3 is optimized for carrying laser light.

More than 30 types of connectors are used on the ends of optical fibers, and there is no real rhyme or reason as to which connectors are used where. The connectors you need to use in a particular case will most often be dictated by your equipment vendors or by your existing building fiber plant. The good news is that conversion jumpers are fairly easy to obtain.
**Ethernet connection and expansion**

Ethernets can be connected through several types of devices. The options below are ranked by approximate cost, with the cheapest options first. The more logic that a device uses to move bits from one network to another, the more hardware and embedded software the device needs to have and the more expensive it is likely to be.

**Hubs**

Devices from a bygone era, hubs are also referred to as concentrators or repeaters. They are active devices that connect Ethernet segments at the physical layer. They require external power.

A hub retimes and retransmits Ethernet frames but does not interpret them; it has no idea where packets are going or what protocol they are using. With the exception of extremely special cases, hubs should no longer be used in enterprise networks; we discourage their use in residential (consumer) networks as well. Switches make significantly more efficient use of network bandwidth and are just as cheap these days.

**Switches**

Switches connect Ethernets at the link layer. They join two physical networks in a way that makes them seem like one big physical network. Switches are the industry standard for connecting Ethernet devices today.

Switches receive, regenerate, and retransmit packets in hardware. Switches use a dynamic learning algorithm. They notice which source addresses come from one port and which from another. They forward packets between ports only when necessary. At first all packets are forwarded, but in a few seconds the switch has learned the locations of most hosts and can be more selective.

Since not all packets are forwarded among networks, each segment of cable that connects to a switch is less saturated with traffic than it would be if all machines were on the same cable. Given that most communication tends to be localized, the increase in apparent bandwidth can be dramatic. And since the logical model of the network is not affected by the presence of a switch, few administrative consequences result from installing one.

Switches can sometimes become confused if your network contains loops. The confusion arises because packets from a single host appear to be on two (or more) ports of the switch. A single Ethernet cannot have loops, but as you connect several Ethernets with routers and switches, the topology can include multiple paths to a host. Some switches can handle this situation by holding alternative routes in reserve in case the primary route goes down. They prune the network they see until the remaining sections present only one path to each node on the network. Some switches can also handle duplicate links between the same two networks and route traffic in a round robin fashion.
Switches must scan every packet to determine if it should be forwarded. Their performance is usually measured by both the packet scanning rate and the packet forwarding rate. Many vendors do not mention packet sizes in the performance figures they quote, and thus, actual performance might be less than advertised.

Although Ethernet switching hardware is getting faster all the time, it is still not a reasonable technology for connecting more than a hundred hosts in a single logical segment. Problems such as “broadcast storms” often plague large switched networks since broadcast traffic must be forwarded to all ports in a switched segment. To solve this problem, use a router to isolate broadcast traffic between switched segments, thereby creating more than one logical Ethernet.

Choosing a switch can be difficult. The switch market is a highly competitive segment of the computer industry, and it’s plagued with marketing claims that aren’t even partially true. When selecting a switch vendor, rely on independent evaluations rather than on data supplied by vendors themselves. In recent years, it has been common for one vendor to have the “best” product for a few months but then completely destroy its performance or reliability when trying to make improvements, thus elevating another manufacturer to the top of the heap.

In all cases, make sure that the backplane speed of the switch is adequate—that’s the number that really counts at the end of a long day. A well-designed switch should have a backplane speed that exceeds the sum of the speeds of all its ports.

**VLAN-capable switches**

Large sites can benefit from switches that partition their ports (through software configuration) into subgroups called virtual local area networks or VLANs. A VLAN is a group of ports that belong to the same logical segment, as if the ports were connected to their own dedicated switch. Such partitioning increases the ability of the switch to isolate traffic, and that capability has beneficial effects on both security and performance.

Traffic among VLANs is handled by a router, or in some cases, by a layer 3 routing module or routing software layer within the switch. An extension of this system known as “VLAN trunking” (such as is specified by the IEEE 802.1Q protocol) allows physically separate switches to service ports on the same logical VLAN.

It’s important to note that VLANs alone provide little additional security. You must filter the traffic among VLANs to reap any potential security benefit.

**Routers**

Routers (aka “layer 3 switches”) direct traffic at the network layer, layer 3 of the OSI network model. They shuttle packets to their final destinations in accordance with the information in the TCP/IP protocol headers. In addition to simply moving packets from one place to another, routers can also perform other functions such as packet filtering (for security), prioritization (for quality of service), and big-picture network topology discovery. See Chapter 15 for all the gory details of how routing actually works.
Routers take one of two forms: fixed configuration and modular.

- Fixed configuration routers have network interfaces permanently installed at the factory. They are usually suitable for small, specialized applications. For example, a router with a T1 interface and an Ethernet interface might be a good choice to connect a small company to the Internet.

- Modular routers have a slot or bus architecture to which interfaces can be added by the end user. Although this approach is usually more expensive, it ensures greater flexibility down the road.

Depending on your reliability needs and expected traffic load, a dedicated router might or might not be cheaper than a UNIX or Linux system configured to act as a router. However, a dedicated router usually achieves superior performance and reliability. This is one area of network design in which it’s usually advisable to spend extra money up front to avoid headaches later.
**Autonegotiation**

With the introduction of a variety of Ethernet standards came the need for devices to figure out how their neighbors were configured and to adjust their settings accordingly. For example, the network won’t work if one side of a link thinks the network is running at 1 Gb/s and the other side of the link thinks it’s running at 10 Mb/s. The Ethernet autonegotiation feature of the IEEE standards is supposed to detect and solve this problem. And in some cases, it does. In other cases, it is easily misapplied and simply compounds the problem.

The two golden rules of autonegotiation are these:

- You *must* use autonegotiation on all interfaces capable of 1 Gb/s or above. It’s required by the standard.
- On interfaces limited to 100 Mb/s or below, you must either configure both ends of a link in autonegotiation mode, or you must *manually* configure the speed and duplex (half vs. full) on both sides. If you configure only one side in autonegotiation mode, that side will not (in most cases) “learn” how the other side has been configured. The result will be a configuration mismatch and poor performance.

To see how to set a network interface’s autonegotiation policy, see the system-specific sections in the *TCP/IP Networking* chapter; they start on this page.
Power over Ethernet

Power over Ethernet (PoE) is an extension of UTP Ethernet (standardized as IEEE 802.3af) that transmits power to devices over the same UTP cable that carries the Ethernet signal. It’s especially handy for Voice over IP (VoIP) telephones or wireless access points (to name just two examples) that need a relatively small amount of power in addition to a network connection.

The power supply capacity of PoE systems has been stratified into four classes that range from 3.84 to 25.5 watts. Never satisfied, the industry is currently working on a higher power standard (802.3bt) that may provide more than 100 watts. Won’t it be convenient to operate an Easy-Bake Oven off the network port in the conference room?

For those of you that are wondering: yes, it is possible to boot a small Linux system off a PoE port. Perhaps the simplest option is a Raspberry Pi with an add-on Pi PoE Switch HAT board.

PoE has two ramifications that are significant for sysadmins:

- You need to be aware of PoE devices in your infrastructure so that you can plan the availability of PoE-capable switch ports accordingly. They are more expensive than non-PoE ports.

- The power budget for data closets that house PoE switches must include the wattage of the PoE devices. Note that you don’t have to budget the same amount of extra cooling for the closet because most of the heat generated by the consumption of PoE power is dissipated outside the closet (usually, in an office).
**Jumbo frames**

Ethernet is standardized for a typical packet size of 1,500 bytes (1,518 with framing), a value chosen long ago when networks were slow and memory for buffers was scarce. Today, these 1,500-byte packets look shrimpy in the context of a gigabit Ethernet. Because every packet consumes overhead and introduces latency, network throughput can be higher if larger packet sizes are allowed.

Unfortunately, the original IEEE standards for the various types of Ethernet forbid large packets because of interoperability concerns. But just as highway traffic often mysteriously flows faster than the stated speed limit, king-size Ethernet packets are a common sight on today’s networks. Egged on by customers, most manufacturers of network equipment have built support for large frames into their gigabit products.

To use these so-called jumbo frames, all you need do is bump up your network interfaces’ MTUs. Throughput gains vary with traffic patterns, but large transfers over TCP (e.g., NFSv4 or SMB file service) benefit the most. Expect a modest but measurable improvement on the order of 10%.

Be aware of these points, though:

- All network equipment on a subnet must support and use jumbo frames, including switches and routers. You cannot mix and match.

- Because jumbo frames are nonstandard, you usually have to enable them explicitly. Devices may accept jumbo frames by default, but they probably will not generate them.

- Since jumbo frames are a form of outlawry, there’s no universal consensus on exactly how large a jumbo frame can or should be. The most common value is 9,000 bytes, or 9,018 bytes with framing. You’ll have to investigate your devices to determine the largest packet size they have in common. Frames larger than 9K or so are sometimes called “super jumbo frames,” but don’t be scared off by the extreme-sounding name. Larger is generally better, at least up to 64K or so.

We endorse the use of jumbo frames on gigabit Ethernets, but be prepared to do some extra debugging if things go wrong. It’s perfectly reasonable to deploy new networks with the default MTU and convert to jumbo frames later once the reliability of the underlying network has been confirmed.
14.2 **WIRELESS: ETHERNET FOR NOMADS**

A wireless network consists of wireless access points (WAPs, or simply APs) and wireless clients. WAPs can be connected to traditional wired networks (the typical configuration) or wirelessly connected to other access points, a configuration known as a “wireless mesh.”
**Wireless standards**

The common wireless standards today are IEEE 802.11g, 802.11n, and 802.11ac. 802.11g operates in the 2.4 GHz frequency band and affords LAN-like access at up to 54 Mb/s. Operating range varies from 100 meters to 40 kilometers, depending on equipment and terrain.

802.11n delivers up to 600 Mb/s of bandwidth and can use both the 5 GHz frequency band and the 2.4 GHz band (though 5 GHz is recommended for deployment). Typical operating range is approximately double that of 802.11g.

The 600 Mb/s bandwidth of 802.11n is largely theoretical. In practice, bandwidth in the neighborhood of 400 Mb/s is a more realistic expectation for an optimized configuration. The environment and capabilities of client devices explain most of the difference between theoretical and real-life throughput. When it comes to wireless, “your mileage may vary” always applies!

802.11ac is an extension of 802.11n with support for up to 1 Gb/s of multistation throughput.

All these standards are covered by the generic term “Wi-Fi.” In theory, the Wi-Fi label is restricted to Ethernet implementations from the IEEE 802.11 family. However, that’s the only kind of wireless Ethernet hardware you can actually buy, so all wireless Ethernet is Wi-Fi.

Today, 802.11g and 802.11n are commonplace. The transceivers are inexpensive and are built into most laptops. Add-in cards are widely and cheaply available for desktop PCs, too.
Wireless client access

You can configure a UNIX or Linux box to connect to a wireless network as a client if you have the right hardware and driver. Since most PC-based wireless cards are still designed for Microsoft Windows, they might not come from the factory with FreeBSD or Linux drivers.

When attempting to add wireless connectivity to a FreeBSD or Linux system, you’ll likely need these commands:

- `ifconfig` – configure a wireless network interface
- `iwlist` – list available wireless access points
- `iwconfig` – configure wireless connection parameters
- `wpa_supplicant` – authenticate to a wireless (or wired 802.1x) network

Unfortunately, the industry’s frantic scramble to sell low-cost hardware often means that getting a wireless adapter to work correctly under UNIX or Linux might require hours of trial and error. Plan ahead, or buy the same adapter that someone else on the Internet has had good luck with on the same OS version you’re running.
Wireless infrastructure and WAPs

Everyone wants wireless everything everywhere, and a wide variety of products are available to provide wireless service. But as with so many things, you get what you pay for. Inexpensive devices often meet the needs of home users but fail to scale well in an enterprise environment.

Wireless topology

WAPs are usually dedicated appliances that consist of one or more radios and some form of embedded network operating system, often a stripped-down version of Linux. A single WAP can provide a connection point for multiple clients, but not for an unlimited number of clients. A good rule of thumb is to serve no more than forty simultaneous clients from a single enterprise-grade WAP. Any device that communicates through a wireless standard supported by your WAPs can act as a client.

WAPs are configured to advertise one or more “service set identifiers,” aka SSIDs. The SSID acts as the name of a wireless LAN and must be unique within a particular area. When a client wants to connect to a wireless LAN, it listens to see what SSIDs are being advertised and lets the user select from among these networks.

You can choose to name your SSID something helpful and easy to remember such as “Third Floor Public,” or you can get creative. Some of our favorite SSID names are

- FBI Surveillance Van
- The Promised LAN
- IP Freely
- Get Off My LAN
- Virus Distribution Center
- Access Denied

Nothing better than geeks at play... In the simplest scenarios, a WAP advertises a single SSID, your client connects to that SSID, and voila! You’re on the network.

However, few aspects of wireless networking are truly simple. What if your house or building is too big to be served by a single WAP? Or what if you need to provide different networks to different groups of users (such as employees vs. guests)? For these cases, you need to strategically structure your wireless network.

You can use multiple SSIDs to break up groups of users or functions. Typically, you map them to separate VLANs, which you can then route or filter as desired, just like wired networks.

The frequency spectrum allocated to 802.11 wireless is broken up into bands, commonly called channels. Left on its own, a WAP selects a quiet radio channel to advertise an SSID. Clients and the WAP then use that channel for communication, forming a single broadcast domain. Nearby
WAPs will likely choose other channels so as to maximize available bandwidth and minimize interference.

The theory is that as clients move around the environment, they will dissociate from one WAP when its signal becomes weak and connect to a closer WAP with a stronger signal. Theory and reality often don’t cooperate, however. Many clients hold onto weak WAP signals with a death grip and ignore better options.

In most situations, you should allow WAPs to automatically select their favorite channels. If you must manually interfere with this process and are using 802.11b/g/n, consider selecting channel 1, 6, or 11. The spectrum allocated to these channels does not overlap, so combinations of these channels create the greatest likelihood of a wide-open wireless highway. The default channels for 802.11a/ac don’t overlap at all, so just pick your favorite number.

Some WAPs have multiple antennas and take advantage of multiple-input, multiple-output technology (MIMO). This practice can increase available bandwidth by exploiting multiple transmitters and receivers to take advantage of signal offsets resulting from propagation delay. The technology can provide a slight performance improvement in some situations, though probably not as much improvement as the dazzling proliferation of antennas might lead you to expect.

If you need a physically larger coverage area, then deploy multiple WAPs. If the area is completely open, you can deploy them in a grid structure. If the physical plant includes walls and other obstructions, you may want to invest in a professional wireless survey. The survey will identify the best options for WAP placement given the physical attributes of your space.

Small money wireless

We like the products made by Ubiquiti (ubnt.com) for inexpensive, high-performing home networks. Google Wifi is a nice, cloud-managed solution, great if you support remote family members. Another option is to run a stripped down version of Linux (such as OpenWrt or LEDE) on a commercial WAP. See openwrt.org for more information and a list of compatible hardware.

Literally dozens of vendors are hawking wireless access points these days. You can buy them at Home Depot and even at the grocery store. El cheapo access points (those in the $30 range) are likely to perform poorly when handling large file transfers or more than one active client.

Big money wireless

Big wireless means big money. Providing reliable, high-density wireless on a large scale (think hospitals, sports arenas, schools, cities) is a challenge complicated by physical plant constraints, user density, and the pesky laws of physics. For situations like this, you need enterprise-grade wireless gear that’s aware of the location and condition of each WAP and that actively adjusts the WAPs’ channels, signal strengths, and client associations to yield the best results. These systems usually support transparent roaming, which allows a client’s association with a particular VLAN and session to seamlessly follow it as the client moves among WAPs.
Our favorite large wireless platforms are those made by Aerohive and Meraki (the latter now owned by Cisco). These next-generation platforms are managed from the cloud, allowing you to sip martinis on the beach as you monitor your network through a browser. You can even eject individual users from the wireless network from the comfort of your beach chair. Take that, hater!

If you are deploying a wireless network on a large scale, you’ll probably need to invest in a wireless network analyzer. We highly recommend the analysis products made by AirMagnet.
Wireless security

The security of wireless networks has traditionally been poor. Wired Equivalent Privacy (WEP) is a protocol used in conjunction with legacy 802.11b networks to encrypt packets traveling over the airwaves. Unfortunately, this standard contains a fatal design flaw that makes it little more than a speed bump for snoopers. Someone sitting outside your building or house can access your network directly and undetectably, usually in under a minute.

More recently, the Wi-Fi Protected Access (WPA) security standards have engendered new confidence in wireless security. Today, WPA (specifically, WPA2) should be used instead of WEP in all new installations. Without WPA2, wireless networks should be considered completely insecure and should never be found inside an enterprise firewall. Don’t even use WEP at home!

To remember that it’s WEP that’s insecure and WPA that’s secure, just remember that WEP stands for Wired Equivalent Privacy. The name is accurate; WEP gives you as much protection as letting someone connect directly to your wired network. (That is, no protection at all—at least at the IP level.)
14.3 SDN: SOFTWARE-DEFINED NETWORKING

Just as with server virtualization, the separation of physical network hardware from the functional architecture of the network can significantly increase flexibility and manageability. The best traction along this path is the software-defined networking (SDN) movement.

The main idea of SDN is that the components managing the network (the control plane) are physically separate from the components that forward packets (the data plane). The data plane is programmable through the control plane, so you can fine-tune or dynamically configure data paths to meet performance, security, and accessibility goals.

As with so many things in our industry, SDN for enterprise networks has become somewhat of a marketing gimmick. The original goal was to standardize vendor-independent ways to reconfigure network components. Although some of this idealism has been realized, many vendors now offer proprietary enterprise SDN products that run somewhat counter to SDN’s original purpose. If you find yourself exploring the enterprise SDN space, choose products that conform to open standards and are interoperable with other vendors’ products.

For large cloud providers, SDN adds a layer of flexibility that reduces your need to know (or care) where a particular resource is physically located. Although these solutions may be proprietary, they are tightly integrated into cloud providers’ platforms and can make configuring your virtual infrastructure effortless.

SDN and its API-driven configuration system offer you, the sysadmin, a tempting opportunity to integrate network topology management with other DevOps-style tools for continuous integration and deployment. Perhaps in some ideal world, you always have a “next at bat” production environment staged and ready to activate with a single click. As the new environment is promoted to production, the network infrastructure magically morphs, eliminating user-visible downtime and the need for you to schedule maintenance windows.
14.4 Network testing and debugging

The key to debugging a network is to break it down into its component parts and then test each piece until you’ve isolated the offending device or cable. The “idiot lights” on switches and hubs (such as “link status” and “packet traffic”) often hold immediate clues to the source of the problem. Top-notch documentation of your wiring scheme is essential for making these indicator lights work in your favor.

As with most tasks, having the right tools for the job is a big part of being able to get the job done right and without delay. The market offers two major types of network debugging tools, although these are quickly converging into unified devices.

The first type of tool is the hand-held cable analyzer. This device can measure the electrical characteristics of a given cable, including its length, with a groovy technology called “time domain reflectrometry.” Usually, these analyzers can also point out simple faults such as broken or miswired cables.

Our favorite product for LAN cable analysis is the Fluke LanMeter. It’s an all-in-one analyzer that can even perform IP pings across the network. High-end versions have their own web server that can show you historical statistics. For WAN (telco) circuits, the T-BERD line analyzer is the cat’s meow. It’s made by Viavi (viavisolutions.com).

The second type of debugging tool is the network sniffer. A sniffer captures the bytes that travel across the wire and disassembles network packets to look for protocol errors, misconfigurations, and general snafus. Sniffers operate at the link layer of the network rather than the electrical layer, so they cannot diagnose cabling problems or electrical issues that might be affecting network interfaces.

Commercial sniffers are available, but we find that the freely available program Wireshark running on a fat laptop is usually the best option. See the Packet sniffers section for details. (Like so many popular programs, Wireshark is often the target of attacks by hackers. Make sure you stay up to date with the most current version.)
14.5 **Building wiring**

If you’re embarking on a building wiring project, the most important advice we can give you is to “do it right the first time.” This is not an area in which to skimp or cut corners. Buying quality materials, selecting a competent wiring contractor, and installing extra connections (drops) will save you years of frustration and heartburn down the road.
UTP cabling options

Category 6a wire typically offers the best price vs. performance tradeoff in today’s market. Its normal format is four pairs per sheath, which is just right for a variety of data connections from RS-232 to gigabit Ethernet.

Category 6a specifications require that the twist be maintained to the point of contact. Special training and termination equipment are necessary to satisfy this requirement. You must use Category 6a jacks and patch panels. We’ve had the best luck with parts manufactured by Siemon.
Connections to offices

For many years, there has been an ongoing debate about how many connections should be wired per office. One connection per office is clearly not enough. But should you use two or four? With the advent of high-bandwidth wireless, we now recommend two, for several reasons:

- A nonzero number of wired connections is typically needed to support voice telephones and other specialty devices.
- Most user devices can now be connected through wireless networking, and users prefer this to being chained down by cables.
- Your network wiring budget is better spent on core infrastructure (fiber to closets, etc.) than on more drops to individual offices.

If you’re in the process of wiring your entire building, you might consider installing a few outlets in hallways, conference rooms, lunch rooms, bathrooms, and of course, ceilings (for wireless access points). Don’t forget to keep security in mind, however, and put publicly accessible ports on a “guest” VLAN that doesn’t have access to your internal network resources. You can also secure public ports by implementing 802.1x authentication.
Wiring standards

Modern buildings often require a large and complex wiring infrastructure to support all the various activities that take place inside. Walking into the average telecommunications closet can be a shocking experience for the weak of stomach, as identically colored, unlabeled wires often cover the walls.

In an effort to increase traceability and standardize building wiring, the Telecommunications Industry Association in February 1993 released TIA/EIA-606 (Administration Standard for Commercial Telecommunications Infrastructure), later updated to TIA/EIA-606-B in 2012.

EIA-606 specifies requirements and guidelines for the identification and documentation of telecommunications infrastructure. Items covered by EIA-606 include

- Termination hardware
- Cables
- Cable pathways
- Equipment spaces
- Infrastructure color coding
- Labeling requirements
- Symbols for standard components

In particular, the standard specifies colors to be used for wiring. Table 14.5 shows the details.

Table 14.5: EIA-606 color chart

<table>
<thead>
<tr>
<th>Termination type</th>
<th>Color</th>
<th>Code</th>
<th>Comments</th>
</tr>
</thead>
<tbody>
<tr>
<td>Demarcation point</td>
<td>Orange</td>
<td>150C</td>
<td>Central office terminations</td>
</tr>
<tr>
<td>Network connections</td>
<td>Green</td>
<td>353C</td>
<td>Also used for aux circuit terminations</td>
</tr>
<tr>
<td>Common equipment b</td>
<td>Purple</td>
<td>264C</td>
<td>Major switching/data eqpt. terminations</td>
</tr>
<tr>
<td>First-level backbone</td>
<td>White</td>
<td>–</td>
<td>Cable terminations</td>
</tr>
<tr>
<td>Second-level backbone</td>
<td>Gray</td>
<td>422C</td>
<td>Cable terminations</td>
</tr>
<tr>
<td>Station</td>
<td>Blue</td>
<td>291C</td>
<td>Horizontal cable terminations</td>
</tr>
<tr>
<td>Interbuilding backbone</td>
<td>Brown</td>
<td>465C</td>
<td>Campus cable terminations</td>
</tr>
<tr>
<td>Miscellaneous</td>
<td>Yellow</td>
<td>101C</td>
<td>Maintenance, alarms, etc.</td>
</tr>
<tr>
<td>Key telephone systems</td>
<td>Red</td>
<td>184C</td>
<td>–</td>
</tr>
</tbody>
</table>

a. Pantone Matching System color code
b. PBXes, hosts, LANs, muxes, etc.

Pantone sells software to map between the Pantone systems for ink-on-paper, textile dyes, and colored plastic. Hey, you could color-coordinate the wiring, the uniforms of the installers, and the wiring documentation! On second thought…
14.6 Network Design Issues

This section addresses the logical and physical design of networks. It’s targeted at medium-sized installations. The ideas presented here scale up to a few hundred hosts but are overkill for three machines and inadequate for thousands. We also assume that you have an adequate budget and are starting from scratch, which is probably only partially true.

Most of network design consists of specifying

- The types of media that will be used
- The topology and routing of cables
- The use of switches and routers

Another key issue in network design is congestion control. For example, file-sharing protocols such as NFS and SMB tax the network quite heavily, and so file serving on a backbone cable is undesirable.

The issues presented in the following sections are typical of those that must be considered in any network design.
Network architecture vs. building architecture

Network architecture is usually more flexible than building architecture, but the two must coexist. If you are lucky enough to be able to specify the network before a building is constructed, be lavish. For most of us, both the building and a facilities-management department already exist and are somewhat rigid.

In existing buildings, the network must use the building architecture, not fight it. Modern buildings often contain utility raceways for data and telephone cables in addition to high-voltage electrical wiring and water or gas pipes. They often use drop ceilings, a boon to network installers. Many campuses and organizations have underground utility tunnels that facilitate network installation.

The integrity of firewalls must be maintained; if you route a cable through a firewall, the hole must be snug and filled in with a noncombustible substance. (This type of firewall is a concrete, brick, or flame-retardant wall that prevents a fire from spreading and burning down the building. Although different from a network security firewall, it’s probably just as important.)

Respect return air plenums in your choice of cable. If you are caught violating fire codes, you might be fined and will be required to fix the problems you have created, even if that means tearing down the entire network and rebuilding it correctly.

Your network’s logical design must fit into the physical constraints of the buildings it serves. As you specify the network, keep in mind that it’s easy to draw a logically good solution and then find that it is physically difficult or impossible to implement.
Expansion

It’s difficult to predict needs ten years into the future, especially in the computer and networking fields. Therefore, design the network with expansion and increased bandwidth in mind. As you install cable, especially in out-of-the-way, hard-to-reach places, pull three to four times the number of pairs you actually need. Remember: the majority of installation cost is labor, not materials.

Even if you have no plans to use fiber, it’s wise to install some when wiring your building, especially in situations where it will be hard to install cable later. Run both multimode and single-mode fiber. The kind you need in the future is always the kind you didn’t install.
**Congestion**

A network is like a chain: it is only as good as its weakest or slowest link. The performance of Ethernet, like that of many other network architectures, degrades nonlinearly as the network becomes loaded.

Overtaxed switches, mismatched interfaces, and low-speed links can all lead to congestion. It’s helpful to isolate local traffic by creating subnets and by using interconnection devices such as routers. Subnets can also be used to cordon off machines that are used for experimentation. It’s difficult to run an experiment that involves several machines if you cannot isolate those machines both physically and logically from the rest of the network.
Maintenance and documentation

We have found that the maintainability of a network correlates highly with the quality of its documentation. Accurate, complete, up-to-date documentation is indispensable.

Cables should be labeled at all termination points. It’s a good idea to post copies of local cable maps inside communications closets so that the maps can be updated on the spot when changes are made. Once every few weeks, have someone copy down the changes for entry into a wiring database.

Joints between major population centers in the form of switches or routers can facilitate debugging by allowing parts of the network to be isolated and debugged separately. It’s also helpful to put joints between political and administrative domains, for similar reasons.
14.7 Management issues

If a network is to work correctly, some things must be centralized, some distributed, and some local. Reasonable ground rules and “good citizen” guidelines must be formulated and agreed on.

A typical environment includes

- A backbone network among buildings
- Departmental subnets connected to the backbone
- Group subnets within a department
- Connections to the outside world (Internet or field office VPNs)

Several facets of network design and implementation must have site-wide control, responsibility, maintenance, and financing. Networks with chargeback algorithms for each connection grow in bizarre but predictable ways as departments try to minimize their own local costs. Prime targets for central control are

- The network design, including the use of subnets, routers, switches, etc.
- The backbone network itself, including the connections to it
- Host IP addresses, hostnames, and subdomain names
- Protocols, mostly to ensure their interoperation
- Routing policy to the Internet

Domain names, IP addresses, and network names are in some sense already controlled centrally by authorities such as ARIN (the American Registry for Internet Numbers) and ICANN. However, your site’s use of these items must be coordinated locally as well.

A central authority has an overall view of the network: its design, capacity, and expected growth. It can afford to own monitoring equipment (and the staff to run it) and to keep the backbone network healthy. It can insist on correct network design, even when that means telling a department to buy a router and build a subnet to connect to the campus backbone. Such a decision might be necessary to ensure that a new connection does not adversely impact the existing network.

If a network serves many types of machines, operating systems, and protocols, it is almost essential to have a layer 3 device as a gateway between networks.
14.8 RECOMMENDED VENDORS

In the past 30+ years of installing networks around the world, we’ve gotten burned more than a few times by products that didn’t quite meet specs or were misrepresented, overpriced, or otherwise failed to meet expectations. Below is a list of vendors in the United States that we still trust, recommend, and use ourselves today.
### Cables and connectors

<table>
<thead>
<tr>
<th>AMP (part of Tyco)</th>
<th>Anixter</th>
<th>Black Box Corporation</th>
</tr>
</thead>
<tbody>
<tr>
<td>(800) 522-6752</td>
<td>(800) 264-9837</td>
<td>(724) 746-5500</td>
</tr>
<tr>
<td>amp.com</td>
<td>anixter.com</td>
<td>blackbox.com</td>
</tr>
<tr>
<td>Belden Cable</td>
<td>Siemon</td>
<td>Newark Electronics</td>
</tr>
<tr>
<td>(800) 235-3361</td>
<td>(860) 945-4395</td>
<td>(800) 463-9275</td>
</tr>
<tr>
<td>(765) 983-5200</td>
<td>siemon.com</td>
<td>newark.com</td>
</tr>
<tr>
<td>belden.com</td>
<td></td>
<td></td>
</tr>
</tbody>
</table>
## Test equipment

<table>
<thead>
<tr>
<th>Fluke</th>
<th>Siemon</th>
<th>Viavi</th>
</tr>
</thead>
<tbody>
<tr>
<td>(800) 443-5853</td>
<td>(860) 945-4395</td>
<td>(844) 468-4284</td>
</tr>
<tr>
<td>fluke.com</td>
<td>siemon.com</td>
<td>viavisolutions.com</td>
</tr>
<tr>
<td>Routers/switches</td>
<td></td>
<td></td>
</tr>
<tr>
<td>-----------------</td>
<td>--</td>
<td></td>
</tr>
<tr>
<td><strong>Cisco Systems</strong></td>
<td><strong>Juniper Networks</strong></td>
<td></td>
</tr>
<tr>
<td>(415) 326-1941</td>
<td>(408) 745-2000</td>
<td></td>
</tr>
<tr>
<td>cisco.com</td>
<td>juniper.net</td>
<td></td>
</tr>
</tbody>
</table>
14.9 **RECOMMENDED READING**


15 IP Routing

More than 4.3 billion IP addresses are available world-wide, so getting packets to the right place on the Internet is no easy task. Chapter 13, TCP/IP Networking, briefly introduced IP packet forwarding. In this chapter, we examine the forwarding process in more detail and investigate several network protocols that allow routers to automatically discover efficient routes. Routing protocols not only lessen the day-to-day administrative burden of maintaining routing information, but they also allow network traffic to be redirected quickly if a router, link, or network should fail.

It’s important to distinguish between the process of actually forwarding IP packets and the management of the routing table that drives this process, both of which are commonly called “routing.” Packet forwarding is simple, whereas route computation is tricky; consequently, the second meaning is used more often in practice. This chapter describes only unicast routing; multicast routing (sending packets to groups of subscribers) involves an array of very different problems and is beyond the scope of this book.

For most cases, the information covered in Chapter 13 is all you need to know about routing. If the appropriate network infrastructure is already in place, you can set up a single static default route (as described in the Routing section) and voilà, you have enough information to reach just about anywhere on the Internet. If you must live within a complex network topology, or if you are using UNIX or Linux systems as part of your network infrastructure, then this chapter’s information about dynamic routing protocols and tools can come in handy. (However, we do not recommend the use of UNIX or Linux systems as network routers in a production infrastructure. Buy a dedicated router.)
IP routing (both for IPv4 and for IPv6) is “next hop” routing. At any given point, the system handling a packet needs to determine only the next host or router in the packet’s journey to its final destination. This is a different approach from that of many legacy protocols, which determine the exact path a packet will travel before it leaves its originating host, a scheme known as source routing. (IP packets can also be source-routed—at least in theory—but this is almost never done. The feature is not widely supported because of security considerations.)
15.1 Packet forwarding: a closer look

Before we jump into the management of routing tables, we need a more detailed look at how the tables are used. Consider the network shown in Exhibit A.

Exhibit A: Example network

For simplicity, we start this example with IPv4; for an IPv6 routing table, see this page.

Router R1 connects two networks, and router R2 connects one of these nets to the outside world. A look at the routing tables for these hosts and routers lets us examine some specific packet forwarding scenarios. First, host A’s routing table:

```
A$ netstat -rn
Destination Gateway Genmask Flags MSS Window ihtt Iface
127.0.0.0 0.0.0.0 255.0.0.0 U 0 0 0 0 lo
199.165.145.0 0.0.0.0 255.255.255.0 U 0 0 0 0 eth0
0.0.0.0 199.165.145.24 0.0.0.0 UG 0 0 0 0 eth0
```

The example above uses the venerable netstat tool to query the routing table. This tool is distributed with FreeBSD and is available for Linux as part of the net-tools package. net-tools is no longer actively maintained, and as a result it is considered deprecated. The less featureful ip route command is the officially recommended way to obtain this information on Linux:

```
A$ ip route
default via 199.165.145.24 dev eth0 onlink
199.165.145.0/24 dev eth0 proto kernel scope link src 199.165.145.17
```

The output from netstat -rn is slightly easier to read, so we use that for subsequent examples and for the following exploration of Exhibit A.

Host A has the simplest routing configuration of the four machines. The first two routes describe the machine’s own network interfaces in standard routing terms. These entries exist so that forwarding to directly connected networks need not be handled as a special case. eth0 is host A’s
Ethernet interface, and lo is the loopback interface, a virtual interface emulated in software. Entries such as these are normally added automatically when a network interface is configured.

See the discussion of netmasks starting on this page.

The default route on host A forwards all packets not addressed to the loopback address or to the 199.165.145 network to the router R1, whose address on this network is 199.165.145.24. Gateways must be only one hop away.

See this page for more information about addressing.

Suppose a process on A sends a packet to B, whose address is 199.165.146.4. The IP implementation looks for a route to the target network, 199.165.146, but none of the routes match. The default route is invoked and the packet is forwarded to R1. Exhibit B shows the packet that actually goes out on the Ethernet. The addresses in the Ethernet header are the MAC addresses of A’s and R1’s interfaces on the 145 net.

Exhibit B: Ethernet packet

<table>
<thead>
<tr>
<th>Ethernet header</th>
<th>IP header</th>
<th>UDP header and data</th>
</tr>
</thead>
<tbody>
<tr>
<td>From: A</td>
<td>From: 199.165.145.17</td>
<td>11001010101010110110101011010110110101</td>
</tr>
<tr>
<td>To: R1</td>
<td>To: 199.165.146.4</td>
<td>0111011011011010110101000101000100</td>
</tr>
<tr>
<td>Type: IP</td>
<td>Type: UDP</td>
<td>01010111011010101010011110101000</td>
</tr>
</tbody>
</table>

The Ethernet destination address is that of router R1, but the IP packet hidden within the Ethernet frame does not mention R1 at all. When R1 inspects the packet it has received, it sees from the IP destination address that it is not the ultimate destination of the packet. It then uses its own routing table to forward the packet to host B without rewriting the IP header; the header still shows the packet coming from A.

Here’s the routing table for host R1:
This table is similar to that of host A, except that it shows two physical network interfaces. The default route in this case points to R2, since that’s the gateway through which the Internet can be reached. Packets bound for either of the 199.165 networks can be delivered directly.

Like host A, host B has only one real network interface. However, B needs an additional route to function correctly because it has direct connections to two different routers. Traffic for the 199.165.145 net must travel through R1, but other traffic should go out to the Internet through R2.

<table>
<thead>
<tr>
<th>Destination</th>
<th>Gateway</th>
<th>Genmask</th>
<th>Flags</th>
<th>MSS</th>
<th>Window</th>
<th>irtt</th>
<th>Iface</th>
</tr>
</thead>
<tbody>
<tr>
<td>127.0.0.0</td>
<td>0.0.0.0</td>
<td>255.0.0.0 U 0 0 0 10</td>
<td></td>
<td></td>
<td></td>
<td></td>
<td></td>
</tr>
<tr>
<td>199.165.145.0</td>
<td>0.0.0.0</td>
<td>255.255.255.0 U 0 0 0 eth0</td>
<td></td>
<td></td>
<td></td>
<td></td>
<td></td>
</tr>
<tr>
<td>199.165.146.0</td>
<td>0.0.0.0</td>
<td>255.255.255.0 U 0 0 0 eth1</td>
<td></td>
<td></td>
<td></td>
<td></td>
<td></td>
</tr>
<tr>
<td>0.0.0.0</td>
<td>199.165.146.3</td>
<td>0.0.0.0</td>
<td>UG 0 0 0 eth1</td>
<td></td>
<td></td>
<td></td>
<td></td>
</tr>
</tbody>
</table>

See [this page](#) for an explanation of ICMP redirects.

In theory, you can configure host B with initial knowledge of only one gateway and rely on help from ICMP redirects to eliminate extra hops. For example, here is one possible initial configuration for host B:

<table>
<thead>
<tr>
<th>Destination</th>
<th>Gateway</th>
<th>Genmask</th>
<th>Flags</th>
<th>MSS</th>
<th>Window</th>
<th>irtt</th>
<th>Iface</th>
</tr>
</thead>
<tbody>
<tr>
<td>127.0.0.0</td>
<td>0.0.0.0</td>
<td>255.0.0.0 U 0 0 0 10</td>
<td></td>
<td></td>
<td></td>
<td></td>
<td></td>
</tr>
<tr>
<td>199.165.145.0</td>
<td>199.165.146.1</td>
<td>255.255.255.0 U 0 0 0 eth0</td>
<td></td>
<td></td>
<td></td>
<td></td>
<td></td>
</tr>
<tr>
<td>199.165.146.0</td>
<td>0.0.0.0</td>
<td>255.255.255.0 U 0 0 0 eth0</td>
<td></td>
<td></td>
<td></td>
<td></td>
<td></td>
</tr>
<tr>
<td>0.0.0.0</td>
<td>199.165.146.3</td>
<td>0.0.0.0</td>
<td>UG 0 0 0 eth0</td>
<td></td>
<td></td>
<td></td>
<td></td>
</tr>
</tbody>
</table>

If B then sends a packet to host A (199.165.145.17), no route matches and the packet is forwarded to R2 for delivery. R2 (which, being a router, presumably has complete information about the network) sends the packet on to R1. Since R1 and B are on the same network, R2 also sends an ICMP redirect notice to B, and B enters a host route for A into its routing table:

| 199.165.145.17 | 199.165.146.1 | 255.255.255.255 | UGHD | 0 | 0 | 0 | eth0 |

This route sends all future traffic for A directly through R1. However, it does not affect routing for other hosts on A’s network, all of which have to be routed by separate redirects from R2.
Some sites use ICMP redirects this way as a sort of low-rent routing “protocol,” thinking that this approach is dynamic. Unfortunately, systems and routers all handle redirects differently. Some hold on to them indefinitely. Others remove them from the routing table after a relatively short period (5–15 minutes). Still others ignore them entirely, which is probably the correct approach from a security perspective.

Redirects have several other potential disadvantages: increased network load, increased load on R2, routing table clutter, and dependence on extra servers, to name a few. Therefore, we don’t recommend their use. In a properly configured network, redirects should never appear in the routing table.

If you are using IPv6 addresses, the same model applies. Here’s a routing table from a FreeBSD host that is running IPv6:

```
$ netstat -rn
Destination Gateway Flags Netif Expire
default 2001:886b:4452::1 UGS re0
2001:886b:4452::/64 link#1 U re0
fe80::/10 ::1 UGRS lo0
fe80::%re0/64 link#1 U re0
```

As in IPv4, the first route is a default that’s used when no more-specific entries match. The next line contains a route to the global IPv6 network where the host lives, 2001:886b:4452::/64. The final two lines are special; they represent a route to the reserved IPv6 network fe80, known as the link-local unicast network. This network is used for traffic that is scoped to the local broadcast domain (typically, the same physical network segment). It is most often used by network services that need to find each other on a unicast network, such as OSPF. Don’t use link-local addresses for normal networking purposes.
15.2 Routing daemons and routing protocols

In simple networks such as the one shown in Exhibit A, it is perfectly reasonable to configure routing by hand. At some point, however, networks become too complicated to be managed this way. Instead of having to explicitly tell every computer on every network how to reach every other computer and network, it would be nice if the computers could just cooperate and figure it all out. This is the job of routing protocols and the daemons that implement them.

Routing protocols have a major advantage over static routing systems in that they can react and adapt to changing network conditions. If a link goes down, then the routing daemons can discover and propagate alternative routes to the networks served by that link, if any such routes exist.

Routing daemons collect information from three sources: configuration files, the existing routing tables, and routing daemons on other systems. This information is merged to compute an optimal set of routes, and the new routes are then fed back into the system routing table (and possibly fed to other systems through a routing protocol). Because network conditions change over time, routing daemons must periodically check in with one another for reassurance that their routing information is still current.

The exact manner in which routes are computed depends on the routing protocol. Two general types of protocols are in common use: distance-vector protocols and link-state protocols.
Distance-vector protocols

Distance-vector (aka “gossipy”) protocols are based on the general idea, “If router X is five hops away from network Y, and I’m adjacent to router X, then I must be six hops away from network Y.” You announce how far you think you are from the networks you know about. If your neighbors don’t know of a better way to get to each network, they mark you as being the best gateway. If they already know a shorter route, they ignore your advertisement. Over time, everyone’s routing tables are supposed to converge to a steady state.

This is a really elegant idea. If it worked as advertised, routing would be relatively simple. Unfortunately, the basic algorithm does not deal well with changes in topology. The problem is that changes in topology can lengthen the optimal routes. Some DV protocols, such as EIGRP, maintain information about multiple possible routes so that they always have a fallback plan. The exact details are not important.

In some cases, infinite loops (e.g., router X receives information from router Y and sends it on to router Z, which sends it back to router Y) can prevent routes from converging at all. Real-world distance-vector protocols must avoid such problems by introducing complex heuristics or by enforcing arbitrary restrictions such as the RIP (Routing Information Protocol) notion that any network more than 15 hops away is unreachable.

Even in nonpathological cases, it can take many update cycles for all routers to reach a steady state. Therefore, to guarantee that routing does not jam for an extended period, the cycle time must be made short, and for this reason distance-vector protocols as a class tend to be talkative. For example, RIP requires that routers broadcast all their routing information every 30 seconds. EIGRP sends updates every 90 seconds.

On the other hand, BGP, the Border Gateway Protocol, transmits the entire table once and then transmits changes as they occur. This optimization substantially reduces the potential for “chatty” (and mostly unnecessary) traffic.

Table 15.1 lists the distance-vector protocols in common use today.

Table 15.1: Common distance-vector routing protocols

<table>
<thead>
<tr>
<th>Name</th>
<th>Long name</th>
<th>Application</th>
</tr>
</thead>
<tbody>
<tr>
<td>RIP</td>
<td>Routing Information Protocol</td>
<td>Internal LANs (if that)</td>
</tr>
<tr>
<td>RIPng</td>
<td>Routing Information Protocol, next generation</td>
<td>IPv6 LANs</td>
</tr>
<tr>
<td>EIGRP</td>
<td>Enhanced Interior Gateway Routing Protocol</td>
<td>WANs, corporate LANs</td>
</tr>
<tr>
<td>BGP</td>
<td>Border Gateway Protocol</td>
<td>Internet backbone routing</td>
</tr>
</tbody>
</table>

a. This protocol (EIGRP) is proprietary to Cisco.
Link-state protocols

Link-state protocols distribute information in a relatively unprocessed form. The records traded among routers are of the form “Router X is adjacent to router Y, and the link is up.” A complete set of such records forms a connectivity map of the network from which each router can compute its own routing table. The primary advantage that link-state protocols offer over distance-vector protocols is the ability to quickly converge on an operational routing solution after a catastrophe occurs. The tradeoff is that maintaining a complete map of the network at each node requires memory and CPU power that would not be needed by a distance-vector routing system.

Because the communications among routers in a link-state protocol are not part of the actual route-computation algorithm, they can be implemented in such a way that transmission loops do not occur. Updates to the topology database propagate across the network efficiently, at a lower cost in network bandwidth and CPU time.

Link-state protocols tend to be more complicated than distance-vector protocols, but this complexity can be explained in part by the fact that link-state protocols make it easier to implement advanced features such as type-of-service routing and multiple routes to the same destination.

The only true link-state protocol in general use is OSPF.
Cost metrics

For a routing protocol to determine which path to a network is shortest, the protocol has to define what is meant by “shortest.” Is it the path involving the fewest number of hops? The path with the lowest latency? The largest minimal intermediate bandwidth? The lowest financial cost?

For routing, the quality of a link is represented by a number called the cost metric. A path cost is the sum of the costs of each link in the path. In the simplest systems, every link has a cost of 1, leading to hop counts as a path metric. But any of the considerations mentioned above can be converted to a numeric cost metric.

Routing protocol designers have labored long and hard to make the definition of cost metrics flexible, and some protocols even allow different metrics to be used for different kinds of network traffic. Nevertheless, in 99% of cases, all this hard work can be safely ignored. The default metrics for most systems work just fine.

You might encounter situations in which the actual shortest path to a destination is not a good default route for political or financial reasons. To handle these cases, you can artificially boost the cost of the critical links to make them seem less appealing. Leave the rest of the routing configuration alone.
**Interior and exterior protocols**

An “autonomous system” (AS) is a group of networks under the administrative control of a single entity. The definition is vague; real-world autonomous systems can be as large as a worldwide corporate network or as small as a building or a single academic department. It all depends on how you want to manage routing. The general tendency is to make autonomous systems as large as you can. This convention simplifies administration and makes routing as efficient as possible.

Routing within an autonomous system is somewhat different from routing between autonomous systems. Protocols for routing among ASs (“exterior” protocols) must often handle routes for many networks (e.g., the entire Internet), and they must deal gracefully with the fact that neighboring routers are under other people’s control. Exterior protocols do not reveal the topology inside an autonomous system, so in a sense they can be thought of as a second level of routing hierarchy that deals with collections of nets rather than individual hosts or cables.

In practice, small- and medium-sized sites rarely need to run an exterior protocol unless they are connected to more than one ISP. With multiple ISPs, the easy division of networks into local and Internet domains collapses, and routers must decide which route to the Internet is best for any particular address. (However, that is not to say that *every* router must know this information. Most hosts can stay stupid and route their default packets through an internal gateway that is better informed.)

Although exterior protocols are not much different from their interior counterparts, this chapter concentrates on the interior protocols and the daemons that support them. If your site must use an external protocol as well, see the recommended reading list [here](#) for some suggested references.
15.3 Protocols on Parade

Several routing protocols are in common use. In this section, we introduce the major players and summarize their main advantages and weaknesses.
RIP and RIPng: Routing Information Protocol

RIP is an old Xerox protocol that was adapted for IP networks. The IP version was originally specified in RFC1058, circa 1988. The protocol has existed in three versions: RIP, RIPv2, and the IPv6-only RIPng (“next generation”).

All versions of RIP are simple distance-vector protocols that use hop counts as a cost metric. Because RIP was designed in an era when computers were expensive and networks small, RIPv1 considers any host 15 or more hops away to be unreachable. Later versions of RIP have maintained the hop-count limit, mostly to encourage the administrators of complex sites to migrate to more sophisticated routing protocols.

See this page for more information about CIDR.

RIPv2 is a minor revision of RIP that distributes netmasks along with next-hop addresses, so its support for subnetted networks and CIDR is better than that of RIPv1. A vague gesture toward increasing the security of RIP was also included.

RIPv2 can be run in a compatibility mode that preserves most of its new features without entirely abandoning vanilla RIP receivers. In most respects, RIPv2 is identical to the original protocol and should be used in preference to it.

See this page for details on IPv6.

RIPng is a restatement of RIP in terms of IPv6. It is an IPv6-only protocol, and RIP remains IPv4-only. If you want to route both IPv4 and IPv6 with RIP, you’ll need to run RIP and RIPng as separate protocols.

Although RIP is known for its profligate broadcasting, it does a good job when a network is changing often or when the topology of remote networks is not known. However, it can be slow to stabilize after a link goes down.

It was originally thought that the advent of more sophisticated routing protocols such as OSPF would make RIP obsolete. However, RIP continues to fill a need for a simple, easy-to-implement protocol that doesn’t require much configuration, and it works well on low-complexity networks.

RIP is widely implemented on non-UNIX platforms. A variety of common devices, from printers to SNMP-manageable network components, can listen to RIP advertisements to learn about network gateways. In addition, some form of RIP client is available for all versions of UNIX and Linux, so RIP is a de facto lowest-common-denominator routing protocol. Often, RIP is used for LAN routing, and a more featureful protocol is used for wide-area connectivity.
Some sites run passive RIP daemons (usually **routed** or Quagga’s **ripd**) that listen for routing updates on the network but do not broadcast any information of their own. The actual route computations are performed with a more efficient protocol such as OSPF (see the next section). RIP is used only as a distribution mechanism.
OSPF: Open Shortest Path First

OSPF is the most popular link-state protocol. “Shortest path first” refers to the mathematical algorithm that calculates routes; “open” is used in the sense of “nonproprietary.” RFC2328 defines the basic protocol (OSPF version 2), and RFC5340 extends it to include support for IPv6 (OSPF version 3). OSPF version 1 is obsolete and is not used.

OSPF is an industrial-strength protocol that works well for large, complicated topologies. It offers several advantages over RIP, including the ability to manage several paths to a single destination and the ability to partition the network into sections (“areas”) that share only high-level routing information. The protocol itself is complex and hence only worthwhile at sites of significant size, where routing protocol behavior really makes a difference. To use OSPF effectively, your site’s IP addressing scheme should be reasonably hierarchical.

The OSPF protocol specification does not mandate any particular cost metric. Cisco’s implementation uses a bandwidth-related value by default.
EIGRP: Enhanced Interior Gateway Routing Protocol

EIGRP is a proprietary routing protocol that runs only on Cisco routers. Its predecessor IGRP was created to address some of the shortcomings of RIP before robust standards like OSPF existed. IGRP has now been deprecated in favor of EIGRP, which accommodates CIDR masks. IGRP and EIGRP are configured similarly despite being quite different in their underlying protocol design.

EIGRP supports IPv6, but as with other routing protocols, the IPv6 world and IPv4 world are configured separately and act as separate, though parallel, routing domains.

EIGRP is a distance-vector protocol, but it’s designed to avoid the looping and convergence problems found in other DV systems. It’s widely regarded as the most evolved distance-vector protocol. For most purposes, EIGRP and OSPF are equally functional.
BGP: Border Gateway Protocol

BGP is an exterior routing protocol; that is, a protocol that manages traffic among autonomous systems rather than among individual networks. There were once several exterior routing protocols in common use, but BGP has outlasted them all.

BGP is now the standard protocol used for Internet backbone routing. As of mid-2017, the Internet routing table contains about 660,000 prefixes. It should be clear from this number that backbone routing has scaling requirements very different from those for local routing.
15.4 **Routing protocol multicast coordination**

Routers need to talk to each other to learn how to get to places on the network, but to get to places on the network they need to talk to a router. This chicken-and-egg problem is most commonly solved through multicast communication. This is the networking equivalent of agreeing to meet your friend on a particular street corner if you get separated. The process is normally invisible to system administrators, but you might occasionally see this multicast traffic in your packet traces or when doing other kinds of network debugging. *Table 15.2* lists the agreed-on multicast addresses for various routing protocols.

<table>
<thead>
<tr>
<th>Description</th>
<th>IPv6</th>
<th>IPv4</th>
</tr>
</thead>
<tbody>
<tr>
<td>All systems on this subnet</td>
<td>ff02::1</td>
<td>224.0.0.1</td>
</tr>
<tr>
<td>All routers on this subnet</td>
<td>ff02::2</td>
<td>224.0.0.2</td>
</tr>
<tr>
<td>Unassigned</td>
<td>ff02::3</td>
<td>224.0.0.3</td>
</tr>
<tr>
<td>DVMRP Routers</td>
<td>ff02::4</td>
<td>224.0.0.4</td>
</tr>
<tr>
<td>OSPF Routers</td>
<td>ff02::5</td>
<td>224.0.0.5</td>
</tr>
<tr>
<td>OSPF DR Routers</td>
<td>ff02::6</td>
<td>224.0.0.6</td>
</tr>
<tr>
<td>RIP Routers</td>
<td>ff02::9</td>
<td>224.0.0.9</td>
</tr>
<tr>
<td>EIGRP Routers</td>
<td>ff02::10</td>
<td>224.0.0.10</td>
</tr>
</tbody>
</table>

*Table 15.2: Routing protocol multicast addresses*
15.5 Routing strategy selection criteria

Routing for a network can be managed at essentially four levels of complexity:

- No routing
- Static routes only
- Mostly static routes, but clients listen for RIP updates
- Dynamic routing everywhere

The topology of the overall network has a dramatic effect on each individual segment’s routing requirements. Different nets might need very different levels of routing support. The following rules of thumb can help you choose a strategy:

- A stand-alone network requires no routing.
- If a network has only one way out, clients (nongateway machines) on that network should have a static default route to the lone gateway. No other configuration is necessary, except perhaps on the gateway itself.
- A gateway with a small number of networks on one side and a gateway to “the world” on the other side can have explicit static routes pointing to the former and a default route to the latter. However, dynamic routing is advisable if both sides have more than one routing choice.
- If networks cross political or administrative boundaries, use dynamic routing at those points, even if the complexity of the networks involved would not otherwise suggest the use of a routing protocol.
- RIP works OK and is widely supported. Don’t reject it out of hand just because it’s an older protocol with a reputation for chattiness.

The problem with RIP is that it doesn’t scale indefinitely; an expanding network will eventually outgrow it. That fact makes RIP something of a transitional protocol with a narrow zone of applicability. That zone is bounded on one side by networks too simple to require any routing protocol and on the other side by networks too complicated for RIP. If your network plans include continued growth, it’s probably reasonable to skip over the “RIP zone” entirely.

- Even when RIP isn’t a good choice for your global routing strategy, it’s still a good way to distribute routes to leaf nodes. But don’t use it where it’s not needed: systems on a network that has only one gateway never need dynamic updates.
- EIGRP and OSPF are about equally functional, but EIGRP is proprietary to Cisco. Cisco makes excellent and cost-competitive routers; nevertheless, standardizing on EIGRP limits your choices for future expansion.

- Routers connected to the Internet through multiple upstream providers must use BGP. However, most routers have only one upstream path and can therefore use a simple static default route.

A good default strategy for a medium-sized site with a relatively stable local structure and a connection to someone else’s net is to use a combination of static and dynamic routing. Routers within the local structure that do not lead to external networks can use static routing, forwarding all unknown packets to a default machine that understands the outside world and does dynamic routing.

A network that is too complicated to be managed with this scheme should rely on dynamic routing. Default static routes can still be used on leaf nets, but machines on networks with more than one router should run routed or some other RIP receiver in passive mode.
15.6 Routing daemons

You should not use UNIX and Linux systems as routers for production networks. Dedicated routers are simpler, more reliable, more secure, and faster (even if they are secretly running a Linux kernel). That said, it’s nice to be able to set up a new subnet with only a $6 network card and a $20 switch. That’s a reasonable approach for lightly populated test and auxiliary networks.

Systems that act as gateways to such subnets don’t need any help managing their own routing tables. Static routes are perfectly adequate, both for the gateway machine and for the machines on the subnet itself. However, if you want the subnet to be reachable by other systems at your site, you need to advertise the subnet’s existence and to identify the router to which packets bound for that subnet should be sent. The usual way to do this is to run a routing daemon on the gateway.

UNIX and Linux systems can participate in most routing protocols through various routing daemons. The notable exception is EIGRP, which, as far as we are aware, has no widely available UNIX or Linux implementation.

Because routing daemons are uncommon on production systems, we don’t describe their use and configuration in detail. However, the following sections outline the common software options and point to detailed configuration information.
**routed: obsolete RIP implementation**

*routed* was for a long time the only standard routing daemon, and it’s still included on a few systems. *routed* speaks only RIP, and poorly at that: even support for RIPv2 is scattershot. *routed* does not speak RIPng, implementation of that protocol being confined to modern daemons such as Quagga.

Where available, *routed* is useful chiefly for its “quiet” mode (`-q`), in which it listens for routing updates but does not broadcast any information of its own. Aside from the command-line flag, *routed* normally does not require configuration. It’s an easy and cheap way to get routing updates without having to deal with much configuration hassle.

---

See [this page](#) for more about manual maintenance of routing tables.

---

*routed* adds its discovered routes to the kernel’s routing table. Routes must be reheard at least every four minutes or they will be removed. However, *routed* knows which routes it has added and does not remove static routes that were installed with the *route* or *ip* commands.
Quagga: mainstream routing daemon

Quagga (quagga.net) is a development fork of Zebra, a GNU project started by Kunihiro Ishiguro and Yoshinari Yoshikawa to implement multiprotocol routing with a collection of independent daemons instead of a single monolithic application. In real life, the quagga—a subspecies of zebra last photographed in 1870—is extinct, but in the digital realm it is Quagga that survives and Zebra that is no longer under active development.

Quagga currently implements RIP (all versions), OSPF (versions 2 and 3), and BGP. It runs on Linux, FreeBSD, and several other platforms. Quagga is either installed by default or is available as an optional package through the system’s standard software repository.

In the Quagga system, the core zebra daemon acts as a central clearing-house for routing information. It manages the interaction between the kernel’s routing table and the daemons for individual routing protocols (ripd, ripngd, ospfd, ospf6d, and bgpd). It also controls the flow of routing information among protocols. Each daemon has its own configuration file in the /etc/quagga directory.

You can connect to any of the Quagga daemons through a command-line interface (vtysh) to query and modify its configuration. The command language itself is designed to be familiar to users of Cisco’s IOS operating system; see the section on Cisco routers below for some additional details. As in IOS, you use enable to enter “superuser” mode, config term to enter configuration commands, and write to save your configuration changes back to the daemon’s configuration file.

The official documentation at quagga.net is available in HTML or PDF form. Although complete, it’s for the most part a workmanlike catalog of options and does not provide much of an overview of the system. The real documentation action is at quagga.net/docs. Look there for well-commented example configurations, FAQs, and tips.

Although the configuration files have a simple format, you’ll need to understand the protocols you’re configuring and have some idea of which options you want to enable or configure. See the recommended reading list here for some good books on routing protocols.
XORP: router in a box

XORP, the eXtensible Open Router Platform project, was started at around the same time as Zebra, but its ambitions are more general. Instead of focusing on routing, XORP aims to emulate all the functions of a dedicated router, including packet filtering and traffic management. Check it out at xorp.org.

One interesting aspect of XORP is that in addition to running under several operating systems (Linux, FreeBSD, macOS, and Windows Server), it’s also available as a live CD that runs directly on PC hardware. The live CD is secretly based on Linux, but it does go a long way toward turning a generic PC into a dedicated routing appliance.
15.7 **Cisco routers**

Routers made by Cisco Systems, Inc., are the de facto standard for Internet routing today. Having captured over 56% of the router market, Cisco’s products are well known, and staff that know how to operate them are relatively easy to find. Before Cisco, UNIX boxes with multiple network interfaces were often used as routers. Today, dedicated routers are the favored gear to put in datacom closets and above ceiling tiles where network cables come together.

Most of Cisco’s router products run an operating system called Cisco IOS, which is proprietary and unrelated to UNIX. Its command set is rather large; the full documentation set fills up about 4.5 feet of shelf space. We could never fully cover Cisco IOS here, but knowing a few basics can get you a long way.

By default, IOS defines two levels of access (user and privileged), both of which are password protected. By default, you can simply `ssh` to a Cisco router to enter user mode.

You are prompted for the user-level access password:

```
$ ssh acme-gw.acme.com
Password: <password>
```

Upon entering the correct password, you receive a prompt from Cisco’s EXEC command interpreter.

```
acme-gw.acme.com>
```

At this prompt, you can enter commands such as `show interfaces` to see the router’s network interfaces or `show ?` to list the other things you can see.

To enter privileged mode, type `enable` and when asked, type the privileged password. Once you have reached the privileged level, your prompt ends in a #:

```
acme-gw.acme.com#
```

Be careful—you can do anything from this prompt, including erasing the router’s configuration information and its operating system. When in doubt, consult Cisco’s manuals or one of the comprehensive books published by Cisco Press.

You can type `show running` to see the current running configuration of the router and `show config` to see the current nonvolatile configuration. Most of the time, these are the same.

Here’s a typical configuration:
The router configuration can be modified in a variety of ways. Cisco offers graphical tools that run under some versions of UNIX/Linux and Windows. Real network administrators never use these; the command prompt is always the sure bet. You can also `scp` a config file to or from a router so you can edit it with your favorite editor.

To modify the configuration from the command prompt, type `config term`.

```
acme-gw.acme.com# config term
Enter configuration commands, one per line. End with CNTL/Z.
acme-gw(config)#
```

You can then type new configuration commands exactly as you want them to appear in the `show running` output. For example, if you wanted to change the IP address of the Ethernet0 interface in the configuration above, you could enter

```
interface Ethernet0
ip address 192.225.40.253 255.255.255.0
```

When you’ve finished entering configuration commands, press <Control-Z> to return to the regular command prompt. If you’re happy with the new configuration, enter `write mem` to save the configuration to nonvolatile memory.
Here are some tips for a successful Cisco router experience:

- Name the router with the **hostname** command. This precaution helps prevent accidents caused by configuration changes to the wrong router. The hostname always appears in the command prompt.

- Always keep a backup router configuration on hand. You can **scp** or **tftp** the running configuration to another system each night for safekeeping.

- It’s often possible to store a copy of the configuration in NVRAM or on a removable jump drive. Do so!

- Control access to the router command line by putting access lists on the router’s VTYs (VTYs are like PTYs on a UNIX system). This precaution prevents unwanted parties from trying to break into your router.

- Control the traffic flowing through your networks (and possibly to the outside world) by setting up access lists on each router interface.

- Keep routers physically secure. It’s easy to reset the privileged password if you have physical access to a Cisco box.

If you have multiple routers and multiple router wranglers, check out the free tool RANCID from shrubbery.net. With a name like RANCID it practically markets itself, but here’s the elevator pitch: RANCID logs into your routers every night to retrieve their configuration files. It diffs the configurations and lets you know about anything that’s changed. It also automatically keeps the configuration files under revision control (see [this page](#)).
15.8 RECOMMENDED READING

Perlman, Radia. *Interconnections: Bridges, Routers, Switches, and Internetworking Protocols (2nd Edition)*. Reading, MA: Addison-Wesley, 2000. This is the definitive work in this topic area. If you buy just one book about networking fundamentals, this should be it. Also, don’t ever pass up a chance to hang out with Radia—she’s a lot of fun and holds a shocking amount of knowledge in her brain.


Huitema, Christian. *Routing in the Internet (2nd Edition)*. Upper Saddle River, NJ: Prentice Hall PTR, 2000. This book is a clear and well-written introduction to routing from the ground up. It covers most of the protocols in common use and also some advanced topics such as multicasting.

There are many routing-related RFCs. Table 15.3 shows the main ones.

<table>
<thead>
<tr>
<th>RFC</th>
<th>Title</th>
<th>Authors</th>
</tr>
</thead>
<tbody>
<tr>
<td>1256</td>
<td>ICMP Router Discovery Messages</td>
<td>Deering</td>
</tr>
<tr>
<td>1724</td>
<td>RIP Version 2 MIB Extension</td>
<td>Malkin, Baker</td>
</tr>
<tr>
<td>2080</td>
<td>RIPng for IPv6</td>
<td>Malkin, Minnear</td>
</tr>
<tr>
<td>2328</td>
<td>OSPF Version 2</td>
<td>Moy</td>
</tr>
<tr>
<td>2453</td>
<td>Routing Information Protocol Version 2</td>
<td>Malkin</td>
</tr>
<tr>
<td>4552</td>
<td>Authentication/Confidentiality for OSPFv3</td>
<td>Gupta, Melam</td>
</tr>
<tr>
<td>4822</td>
<td>RIPv2 Cryptographic Authentication</td>
<td>Atkinson, Fanto</td>
</tr>
<tr>
<td>4861</td>
<td>Neighbor Discovery for IPv6</td>
<td>Narten et al.</td>
</tr>
<tr>
<td>5175</td>
<td>IPv6 Router Advertisement Flags Option</td>
<td>Haberman, Hinden</td>
</tr>
<tr>
<td>5308</td>
<td>Routing IPv6 with IS-IS</td>
<td>Hopps</td>
</tr>
<tr>
<td>5340</td>
<td>OSPF for IPv6</td>
<td>Coltun et al.</td>
</tr>
<tr>
<td>5643</td>
<td>Management Information Base for OSPFv3</td>
<td>Joyal, Manral, et al.</td>
</tr>
</tbody>
</table>
16 DNS: The Domain Name System

The Internet delivers instant access to resources all over the world, and each of those computers or sites has a unique name (e.g., google.com). However, anyone who has tried to find a friend or a lost child in a crowded stadium knows that simply knowing a name and yelling it loudly is not enough. Essential to finding anything (or anyone) is an organized system for communicating, updating, and distributing names and their locations.

Users and user-level programs like to refer to resources by name (e.g., amazon.com), but low-level network software understands only IP addresses (e.g., 54.239.17.6). Mapping between names and addresses is the best known and arguably most important function of DNS, the Domain Name System. DNS includes other elements and features, but almost without exception they exist to support this primary objective.

Over the history of the Internet, DNS has been both praised and criticized. Its initial elegance and simplicity encouraged adoption in the early years and enabled the Internet to grow quickly with little centralized management. As needs for additional functionality grew, so did the DNS system. Sometimes, these functions were bolted on in a way that looks ugly today. Naysayers point out weaknesses in the DNS infrastructure as evidence that the Internet is on the verge of collapse.

Say what you will, but the fundamental concepts and protocols of DNS have so far withstood growth from a few hundred hosts in a single country to a world-wide network that supports over 3 billion users across more than 1 billion hosts. Nowhere else can we find an information system
that has grown to this scale with so few issues. Without DNS, the Internet would have failed long ago.
16.1 DNS Architecture

DNS is a distributed database. Under this model, one site stores the data for computers it knows about, another site stores the data for its own set of computers, and the sites cooperate and share data when one site needs to look up the other’s data. From an administrative point of view, the DNS servers you have configured for your domain answer queries from the outside world about names in your domain; they also query other domains’ servers on behalf of your users.
Queries and responses

A DNS query consists of a name and a record type. The answer returned is a set of “resource records” (RRs) that are responsive to the query (or alternatively, a response indicating that the name and record type you asked for do not exist).

“Responsive” doesn’t necessarily mean “dispositive.” DNS servers are arranged into a hierarchy, and it might be necessary to contact servers at several layers to answer a particular query (see this page). Servers that don’t know the answer to a query return resource records that help the client locate a server that does.

The most common query is for an A record, which returns the IP address associated with a name. Exhibit A illustrates a typical scenario.

Exhibit A: A simple name lookup

First, a human types the name of a desired site into a web browser. The browser then calls the DNS “resolver” library to look up the corresponding address. The resolver library constructs a query for an A record and sends it to a name server, which returns the A record in its response. Finally, the browser opens a TCP connection to the target host through the IP address returned by the name server.
DNS service providers

Years ago, one of the core tasks of every system administrator was to set up and maintain a DNS server for their organization. Today, the landscape has changed. If an organization maintains a DNS server at all, it is frequently for internal use only.

Microsoft’s Active Directory system includes an integrated DNS server that meshes nicely with the other Microsoft-flavored services found in corporate environments. However, Active Directory is suitable only for internal use. It should never be used as an external (Internet-facing) DNS server because of potential security concerns.

Every organization still needs an external-facing DNS server, but it’s now common to use one of the many commercial “managed” DNS providers for this function. These services offer a GUI management interface and highly available, secure DNS infrastructure for only pennies (or dollars) a day. Amazon Route 53, CloudFlare, GoDaddy, DNS Made Easy, and Rackspace are just a few of the major providers.

Of course, you can still set up and maintain your own DNS server (internal or external) if you wish. You have dozens of DNS implementations to choose from, but the Berkeley Internet Name Domain (BIND) system still dominates the Internet. Over 75% of DNS servers run some form of it, according to the July, 2015 ISC Internet Domain Survey.

Regardless of which path you choose, as a system administrator you need to understand the basic concepts and architecture of DNS. The first few sections of this chapter focus on that important foundational knowledge. Starting on this page, we show some specific configurations for BIND.
16.2 DNS FOR LOOKUPS

Regardless of whether you run your own name server, use a managed DNS service, or have someone else providing DNS service for you, you’ll certainly want to configure all of your systems to look up names in DNS.

Two steps are needed to make this happen. First, you configure your systems as DNS clients. Second, you tell the systems when to use DNS as opposed to other name lookup methods such as a static /etc/hosts file.
resolv.conf: client resolver configuration

Each host on the network should be a DNS client. You configure the client-side resolver in the file `/etc/resolv.conf`. This file lists the name servers to which the host can send queries.

See [this page](#) for more information about DHCP.

If your host gets its IP address and network parameters from a DHCP server, the `/etc/resolv.conf` file is normally set up for you automatically. Otherwise, you must edit the file by hand. The format is

```
search domainname ...
nameserver ipaddr
```

Up to three name servers can be listed. Here’s a complete example:

```
search atrust.com booklab.atrust.com
nameserver 63.173.189.1 ; ns1
nameserver 174.129.219.225 ; ns2
```

The `search` line lists the domains to query if a hostname is not fully qualified. For example, if a user issues the command `ssh coraline`, the resolver completes the name with the first domain in the search list and looks for `coraline.atrust.com`. If no such name exists, the resolver also tries `coraline.booklab.atrust.com`. The number of domains that can be specified in a `search` directive is resolver-specific; most allow between six and eight, with a limit of 256 characters.

The name servers listed in `resolv.conf` must be configured to allow your host to submit queries. They must also be recursive; that is, they must answer queries to the best of their ability and not try to refer you to other name servers; see [this page](#).

DNS servers are contacted in order. As long as the first one continues to answer queries, the others are ignored. If a problem occurs, the query eventually times out and the next name server is tried. Each server is tried in turn, up to four times. The timeout interval increases with each failure. The default timeout interval is five seconds, which seems like forever to impatient users.
nsswitch.conf: who do I ask for a name?

Both FreeBSD and Linux use a switch file, /etc/nsswitch.conf, to specify how hostname-to-IP-address mappings should be performed and whether DNS should be tried first, last, or not at all. If no switch file is present, the default behavior is

hosts: dns [!UNAVAIL=return] files

The !UNAVAIL clause means that if DNS is available but a name is not found there, the lookup attempt should fail rather than continuing to the next entry (in this case, the /etc/hosts file). If no name server is running (as might be the case during boot), the lookup process does consult the hosts file.

Our example distributions all provide the following default nsswitch.conf entry:

hosts: files dns

This configuration gives precedence to the /etc/hosts file, which is always checked. DNS is consulted only for names that are unresolvable through /etc/hosts.

There is really no best way to configure lookups—it depends on how your site is managed. In general, we prefer to keep as much host information as possible in DNS but always preserve the ability to fall back to the static hosts file during the boot process if necessary.

If name service is provided for you by an outside organization, you might be done with DNS configuration after setting up resolv.conf and nsswitch.conf. If so, you can skip the rest of this chapter, or read on to learn more.
16.3 The DNS Namespace

The DNS namespace is organized into a tree that contains both forward mappings and reverse mappings. Forward mappings map hostnames to IP addresses (and other records), and reverse mappings map IP addresses to hostnames. Every complete hostname (e.g., nubark.atrust.com) is a node in the forward branch of the tree, and (in theory) every IP address is a node in the reverse branch. Exhibit B shows the general layout of the naming tree.

Exhibit B: DNS zone tree

<table>
<thead>
<tr>
<th>Reverse zones</th>
<th>Forward zones</th>
</tr>
</thead>
<tbody>
<tr>
<td>arpa</td>
<td>. (root)</td>
</tr>
<tr>
<td>in-addr</td>
<td>com</td>
</tr>
<tr>
<td></td>
<td>org</td>
</tr>
<tr>
<td></td>
<td>net</td>
</tr>
<tr>
<td></td>
<td>de</td>
</tr>
<tr>
<td></td>
<td>au</td>
</tr>
<tr>
<td></td>
<td>...</td>
</tr>
<tr>
<td>... 61   62  63 ...</td>
<td>...</td>
</tr>
<tr>
<td>... 173 ...</td>
<td></td>
</tr>
<tr>
<td>... 189 ...</td>
<td></td>
</tr>
<tr>
<td>... 1 ...</td>
<td></td>
</tr>
<tr>
<td></td>
<td>www</td>
</tr>
<tr>
<td></td>
<td>www</td>
</tr>
<tr>
<td></td>
<td>nubark</td>
</tr>
</tbody>
</table>

To allow the same DNS system to manage both names (which have the most significant information on the right), and IP addresses (which have the most significant part on the left), the IP branch of the namespace is inverted by listing the octets of the IP address backwards. For example, if host nubark.atrust.com has IP address 63.173.189.1, the corresponding node of the forward branch of the naming tree is “nubark.atrust.com.” and the node of the reverse branch is “1.189.173.63.in-addr.arpa.”. The in-addr.arpa portion of the name is a fixed suffix.

Both of these names end with a dot, just as the full pathnames of files always start with a slash. That makes them “fully qualified domain names” or FQDNs for short. Outside the context of DNS, names like nubark.atrust.com (without the final dot) are sometimes referred to as “fully qualified hostnames,” but this is a colloquialism. Within the DNS system itself, the presence or absence of the trailing dot is of crucial importance.

Two types of top-level domains exist: country code domains (ccTLDs) and generic top-level domains (gTLDs). ICANN, the Internet Corporation for Assigned Names and Numbers, accredits various agencies to be part of its shared registry project for registering names in the gTLDs such as com, net, and org. To register for a ccTLD name, check the IANA (Internet Assigned Numbers Authority) web page iana.org/cctld to find the registry in charge of a particular country’s registration.
Registering a domain name

To obtain a second-level domain name (such as blazedgoat.com), you must apply to a registrar for the appropriate top-level domain. To complete the domain registration forms, you must choose a name that is not already taken and identify a technical contact person, an administrative contact person, and at least two hosts that will be name servers for your domain. Fees vary among registrars, but these days they are all generally quite inexpensive.
Creating your own subdomains

The procedure for creating a subdomain is similar to that for creating a second-level domain, except that the central authority is now local (or more accurately, within your own organization). Specifically, the steps are as follows:

- Choose a name that is unique in the local context.
- Identify two or more hosts to be servers for your new domain.
- Coordinate with the administrator of the parent domain.

The two-or-more-servers rule is a policy, not a technical requirement. You make the rules in your own subdomains, so you can get away with a single server if you want.

Parent domains should check to be sure that a child domain’s name servers are up and running before performing the delegation. If the servers are not working, a “lame delegation” results, and you might receive nasty email asking you to clean up your DNS act. Lame delegations are covered in more detail here.
16.4 How DNS works

Name servers around the world work together to answer queries. Typically, they distribute information maintained by whichever administrator is closest to the query target. Understanding the roles and relationships of name servers is important both for day-to-day operations and for debugging.
Name servers

A name server performs several chores:

- It answers queries about your site’s hostnames and IP addresses.
- It asks about both local and remote hosts on behalf of your users.
- It caches the answers to queries so that it can answer faster next time.
- It communicates with other local name servers to keep DNS data synchronized.

Name servers deal with “zones,” where a zone is essentially a domain minus its subdomains. You will often see the term “domain” used where a zone is what’s actually meant, even in this book.

Name servers can operate in several different modes. The distinctions among them fall along several axes, so the final categorization is often not tidy. To make things even more confusing, a single server can play different roles with respect to different zones. Table 16.1 lists some of the adjectives used to describe name servers.

Table 16.1: Name server taxonomy

<table>
<thead>
<tr>
<th>Type of server</th>
<th>Description</th>
</tr>
</thead>
<tbody>
<tr>
<td>authoritative</td>
<td>Officially represents a zone</td>
</tr>
<tr>
<td>master</td>
<td>The master server for a zone; gets its data from a disk file</td>
</tr>
<tr>
<td>primary</td>
<td>Another name for the master server</td>
</tr>
<tr>
<td>slave</td>
<td>Copies its data from the master</td>
</tr>
<tr>
<td>secondary</td>
<td>Another name for a slave server</td>
</tr>
<tr>
<td>stub</td>
<td>Like a slave, but copies only name server data (not host data)</td>
</tr>
<tr>
<td>distribution</td>
<td>A server advertised only within a domain (aka “stealth server”)</td>
</tr>
<tr>
<td>nonauthoritative(^a)</td>
<td>Answers a query from cache; doesn’t know if the data is still valid</td>
</tr>
<tr>
<td>caching</td>
<td>Caches data from previous queries; usually has no local zones</td>
</tr>
<tr>
<td>forwarder</td>
<td>Performs queries on behalf of many clients; builds a large cache</td>
</tr>
<tr>
<td>recursive</td>
<td>Queries on your behalf until it returns either an answer or an error</td>
</tr>
<tr>
<td>nonrecursive</td>
<td>Refers you to another server if it can’t answer a query</td>
</tr>
</tbody>
</table>

\(^a\) Strictly speaking, “nonauthoritative” is an attribute of a DNS query response, not a server.

These categorizations vary according to the name server’s source of data (authoritative, caching, master, slave), the type of data saved (stub), the query path (forwarder), the completeness of answers handed out (recursive, nonrecursive), and finally, the visibility of the server (distribution). The next few sections provide additional details on the most important of these distinctions; the others are described elsewhere in this chapter.
Authoritative and caching-only servers

Master, slave, and caching-only servers are distinguished by two characteristics: where the data comes from, and whether the server is authoritative for the domain. Each zone typically has one master name server. The master server keeps the official copy of the zone’s data on disk. The system administrator changes the zone’s data by editing the master server’s data files.

Some sites use multiple masters or even no masters. However, these are unusual configurations. We describe only the single-master case.

See this page for more information about zone transfers.

A slave server gets its data from the master server through a “zone transfer” operation. A zone can have several slave name servers and must have at least one. A stub server is a special kind of slave that loads only the NS (name server) records from the master. It’s fine for the same machine to be both a master server for some zones and a slave server for other zones.

A caching-only name server loads the addresses of the servers for the root domain from a startup file and accumulates the rest of its data by caching answers to the queries it resolves. A caching-only name server has no data of its own and is not authoritative for any zone (except perhaps the localhost zone).

An authoritative answer from a name server is “guaranteed” to be accurate; a nonauthoritative answer might be out of date. However, a very high percentage of nonauthoritative answers are perfectly correct. Master and slave servers are authoritative for their own zones, but not for information they may have cached about other domains. Truth be told, even authoritative answers can be inaccurate if a sysadmin changes the master server’s data but forgets to propagate the changes (e.g., doesn’t change the zone’s serial number).

At least one slave server is required for each zone. Ideally, there should be at least two slaves, one of which is in a location that does not share common infrastructure with the master. On-site slaves should live on different networks and different power circuits. When name service stops, all normal network access stops, too.
Recursive and nonrecursive servers

Name servers are either recursive or nonrecursive. If a nonrecursive server has the answer to a query cached from a previous transaction or is authoritative for the domain to which the query pertains, it provides an appropriate response. Otherwise, instead of returning a real answer, it returns a referral to the authoritative servers of another domain that are more likely to know the answer. A client of a nonrecursive server must be prepared to accept and act on referrals.

Although nonrecursive servers might seem lazy, they usually have good reason not to take on extra work. Authoritative-only servers (e.g., root servers and top-level domain servers) are all nonrecursive, but since they may process tens of thousands of queries per second we can excuse them for cutting corners.

A recursive server returns only real answers and error messages. It follows referrals itself, relieving clients of this responsibility. In other respects, the basic procedure for resolving a query is essentially the same.

For security, an organization’s externally accessible name servers should always be nonrecursive. Recursive name servers that are visible to the world can be vulnerable to cache poisoning attacks.

Note well: resolver libraries do not understand referrals. Any local name server listed in a client’s resolv.conf file must be recursive.
Resource records

Each site maintains one or more pieces of the distributed database that makes up the world-wide DNS system. Your piece of the database consists of text files that contain records for each of your hosts; these are known as “resource records.” Each record is a single line consisting of a name (usually a hostname), a record type, and some data values. The name field can be omitted if its value is the same as that of the previous line.

For example, the lines

<table>
<thead>
<tr>
<th>nubark</th>
<th>IN</th>
<th>A</th>
<th>63.173.189.1</th>
</tr>
</thead>
<tbody>
<tr>
<td></td>
<td>IN</td>
<td>MX</td>
<td>mailserver.atrust.com</td>
</tr>
</tbody>
</table>

in the “forward” file (called atrust.com), and the line

| 1          | IN | PTR | nubark.atrust.com |

See this page for more information about MX records.

in the “reverse” file (called 63.173.189.rev) associate nubark.atrust.com with the IP address 63.173.189.1. The MX record routes email addressed to this machine to the host mailserver.atrust.com.

The IN fields denote the record classes. In practice, this field is always IN for Internet.

Resource records are the lingua franca of DNS and are independent of the configuration files that control the operation of any given DNS server implementation. They are also the pieces of data that flow around the DNS system and become cached at various locations.
Delegation

All name servers read the identities of the root servers from a local config file or have them built into the code. The root servers know the name servers for com, net, edu, fi, de, and other top-level domains. Farther down the chain, edu knows about colorado.edu, berkeley.edu, and so on. Each domain can delegate authority for its subdomains to other servers.

Let’s inspect a real example. Suppose we want to look up the address for the machine vangogh.cs.berkeley.edu from the machine lair.cs.colorado.edu. The host lair asks its local name server, ns.cs.colorado.edu, to figure out the answer. The following illustration (Exhibit C) shows the subsequent events.

Exhibit C: DNS query process for vangogh.cs.berkeley.edu

The numbers on the arrows between servers show the order of events, and a letter denotes the type of transaction (query, referral, or answer). We assume that none of the required information was cached before the query, except for the names and IP addresses of the servers of the root domain.

The local server doesn’t know vangogh’s address. In fact, it doesn’t know anything about cs.berkeley.edu or berkeley.edu or even edu. It does know servers for the root domain, however, so it queries a root server about vangogh.cs.berkeley.edu and receives a referral to the servers for edu.

The local name server is a recursive server. When the answer to a query consists of a referral to another server, the local server resubmits the query to the new server. It continues to follow referrals until it finds a server that has the data it’s looking for.

In this case, the local name server sends its query to a server of the edu domain (asking, as always, about vangogh.cs.berkeley.edu) and gets back a referral to the servers for berkeley.edu. The local name server then repeats this same query on a berkeley.edu server. If the Berkeley server doesn’t have the answer cached, it returns a referral to the servers for cs.berkeley.edu. The
cs.berkeley.edu server is authoritative for the requested information, looks the answer up in its zone files, and returns vangogh’s address.

When the dust settles, ns.cs.colorado.edu has cached vangogh’s address. It has also cached data on the servers for edu, berkeley.edu, and cs.berkeley.edu.

You can view the query process in detail with `dig +trace` or `drill -T`. (`dig` and `drill` are DNS query tools: `dig` from the BIND distribution and `drill` from NLnet Labs.)
Caching and efficiency

Caching increases the efficiency of lookups: a cached answer is almost free and is usually correct because hostname-to-address mappings change infrequently. An answer is saved for a period of time called the “time to live” (TTL), which is specified by the owner of the data record in question.

Most queries are for local hosts and can be resolved quickly. Users also inadvertently help with efficiency because they repeat many queries; after the first instance of a query, the repeats are more or less free.

Under normal conditions, your site’s resource records should use a TTL that is somewhere between an hour and a day. The longer the TTL, the less network traffic will be consumed by Internet clients obtaining fresh copies of the record.

If you have a specific service that is load-balanced across logical subnets (often called “global server load balancing”), you may be required by your load-balancing vendor to choose a shorter TTL, such as 10 seconds or 1 minute. The short TTL lets the load balancer react quickly to inoperative servers and denial of service attacks. The system still works correctly with short TTLs, but your name servers have to work hard.

In the vangogh example above, the TTLs were 42 days for the roots, 2 days for edu, 2 days for berkeley.edu, and 1 day for vangogh.cs.berkeley.edu. These are reasonable values. If you are planning a massive renumbering, change the TTLs to a shorter value well before you start.

DNS servers also implement negative caching. That is, they remember when a query fails and do not repeat that query until the negative caching TTL value has expired. Negative caching can potentially save answers of the following types:

- No host or domain matches the name queried.
- The type of data requested does not exist for this host.
- The server is not responding.
- The server is unreachable because of network problems.

The BIND implementation caches the first two types of negative data and allows the negative cache times to be configured.
Multiple answers and round robin DNS load balancing

A name server often receives multiple records in response to a query. For example, the response to a query for the name servers of the root domain would list all 13 servers.

You can take advantage of this balancing effect for your own servers by assigning several different IP addresses (for different machines) to a single hostname:

```
www IN A 192.168.0.1
IN A 192.168.0.2
IN A 192.168.0.3
```

Most name servers return multirecord sets in a different order each time they receive a query, rotating them in round robin fashion. When a client receives a response with multiple records, the most common behavior is to try the addresses in the order returned by the DNS server. However, this behavior is not required. Some clients may behave differently.

This scheme is commonly referred to as round robin DNS load balancing. However, it is a crude solution at best. Large sites use load-balancing software (such as HAProxy; see [this page](#)) or dedicated load-balancing appliances.
Debugging with query tools

See this page for more information about DNSSEC.

Five command-line tools that query the DNS database are distributed with BIND: nslookup, dig, host, drill and delv. nslookup and host are simple and have pretty output, but you need dig or drill to get all the details. drill is better for following DNSSEC signature chains. The name drill is a pun on dig (the Domain Information Groper), implying you can get even more info from DNS with drill than you can with dig. delv is new to BIND 9.10 and will eventually replace drill for DNSSEC debugging.

See this page for more information about split DNS.

By default, dig and drill query the name servers configured in /etc/resolv.conf. A @nameserver argument makes either command query a specific name server. The ability to query a particular server lets you check to be sure that any changes you make to a zone have been propagated to secondary servers and to the outside world. This feature is especially useful if you use views (split DNS) and need to verify that you have configured them correctly.

If you specify a record type, dig and drill query for that type only. The pseudo-type any is a bit sneaky: instead of returning all data associated with a name, it returns all cached data associated with the name. So, to get all records, you might have to do dig domain NS followed by dig @ns1.domain domain any. (Authoritative data counts as cached in this context.)

dig has about 50 options and drill about half that many. Either command accepts an -h flag to list the various options. (You’ll probably want to pipe the output through less.) For both tools, -x reverses the bytes of an IP address and does a reverse query. The +trace flag to dig or -T to drill shows the iterative steps in the resolution process from the roots down.

dig and drill include the notation aa in the output flags if an answer is authoritative (i.e., it comes directly from a master or slave server of that zone). The code ad indicates that an answer was authenticated by DNSSEC. When testing a new configuration, be sure that you look up data for both local and remote hosts. If you can access a host by IP address but not by name, DNS is probably the culprit.

The most common use of dig is to determine what records are currently being returned for a particular name. If only an AUTHORITY response is returned, you have been referred to another name server. If an ANSWER response is returned, your question has been directly answered (and other information may be included as well).

It’s often useful to follow the delegation chain manually from the root servers to verify that everything is in the right place. Below we look at an example of that process for the name
www.viawest.com. First, we query a root server to see who is authoritative for viawest.com by requesting the start-of-authority (SOA) record:

```
$ dig @a.root-servers.net viawest.com soa
;; DiG 9.8.3-P1 <<>> @a.root-servers.net viawest.com soa
;; global options: +cmd
;; Got answer:
;; ->>HEADER<<- opcode: QUERY, status: NOERROR, id: 7824
;; flags: qr rd; QUERY: 1, ANSWER: 0, AUTHORITY: 13, ADDITIONAL: 14
;; WARNING: recursion requested but not available

;; QUESTION SECTION:
viawest.com. IN SOA

;; AUTHORITY SECTION:
com. 172800 IN NS c.gtld-servers.net.
com. 172800 IN NS b.gtld-servers.net.
com. 172800 IN NS a.gtld-servers.net.
...

;; ADDITIONAL SECTION:
c.gtld-servers.net. 172800 IN A 192.26.92.30
b.gtld-servers.net. 172800 IN A 192.33.14.30
b.gtld-servers.net. 172800 IN AAAA 2001:503:231d::2:30
a.gtld-servers.net. 172800 IN A 192.5.6.30
...

;; Query time: 62 msec
;; SERVER: 198.41.0.4#53(198.41.0.4)
;; MSG SIZE  rcvd: 489
```

Note that the status returned is **NOERROR**. That tells us that the query returned a response without notable errors. Other common status values are **NXDOMAIN**, which indicates the name requested doesn’t exist (or isn’t registered), and **SERVFAIL**, which usually indicates a configuration error on the name server itself.

This **AUTHORITY SECTION** tells us that the global top-level domain (gTLD) servers are the next link in the authority chain for this domain. So, we pick one at random and repeat the same query:
This response is much more succinct, and we now know that the next server to query is ns1.viawest.com (or ns2.viawest.com).
$ dig @ns1.viawest.net viawest.com soa
 ; <<< DIG 9.8.3-P1 <<< @ns2.viawest.net viawest.com soa
 ; (1 server found)
 ;; global options: +cmd
 ;; Got answer:
 ;; ->>HEADER<<- opcode: QUERY, status: NOERROR, id: 61543
 ;; flags: qr aa rd; QUERY: 1, ANSWER: 1, AUTHORITY: 1, ADDITIONAL: 1
 ;; WARNING: recursion requested but not available

 ;; QUESTION SECTION:
 ;viawest.com. IN SOA

 ;; ANSWER SECTION:
 viawest.com. 3600 IN SOA mvec.viawest.net. hostmaster.
                      viawest.net. 2007112567 3600 1800 1209600 3600

 ;; AUTHORITY SECTION:
 viawest.com. 86400 IN NS ns2.viawest.net.

 ;; ADDITIONAL SECTION:
 ns2.viawest.net. 3600 IN A 209.170.216.2

 ;; Query time: 5 msec
 ;; SERVER: 216.87.64.12#53(216.87.64.12)
 ;; MSG SIZE  rcvd: 126

This query returns an answer for the viawest.com domain. We now know an authoritative name server and can query for the name we actually want, www.viawest.com.
This final query shows us that www.viawest.com has a CNAME record pointed at hm-d8ebfa-via1.threatx.io, meaning that it is another name for the threatx host (a host operated by a cloud-based distributed denial-of-service provider).

Of course, if you query a recursive name server, it will follow the entire delegation chain on your behalf. But when debugging, it’s typically more useful to investigate the chain link by link.
16.5 The DNS Database

A zone’s DNS database is a set of text files maintained by the system administrator on the zone’s master name server. These text files are often called zone files. They contain two types of entries: parser commands (things like $ORIGIN and $TTL) and resource records. Only the resource records are really part of the database; the parser commands just provide some shorthand ways to enter records.
Parser commands in zone files

Zone file commands are standardized in RFCs 1035 and 2308.

Commands can be embedded in zone files to make the zone files more readable and easier to maintain. The commands either influence the way the parser interprets subsequent records or they expand into multiple DNS records themselves. Once a zone file has been read and interpreted, none of these commands remain a part of the zone’s data (at least, not in their original forms).

Three commands ($ORIGIN, $INCLUDE, and $TTL) are standard for all DNS implementations, and a fourth, $GENERATE, is found only in BIND. Commands must start in column one and occur on a line by themselves.

Zone files are read and parsed from top to bottom in a single pass. As the name server reads a zone file, it adds the default domain (or “origin”) to any names that are not already fully qualified. The origin defaults to the domain name specified in the name server’s configuration file. However, you can set the origin or change it within a zone file by using the $ORIGIN directive:

$ORIGIN domain-name

The use of relative names where fully qualified names are expected saves lots of typing and makes zone files much easier to read.

Many sites use the $INCLUDE directive in their zone database files to separate overhead records from data records, to separate logical pieces of a zone file, or to keep cryptographic keys in a file with restricted permissions. The syntax is

$INCLUDE filename [origin]

The specified file is read into the database at the point of the $INCLUDE directive. If filename is not an absolute path, it is interpreted relative to the home directory of the running name server.

If you supply an origin value, the parser acts as if an $ORIGIN directive precedes the contents of the file being read. Watch out: the origin does not revert to its previous value after the $INCLUDE has been executed. You’ll probably want to reset the origin, either at the end of the included file or on the line following the $INCLUDE statement.

The $TTL directive sets a default value for the time-to-live field of the records that follow it. It must be the first line of the zone file. The default units for the $TTL value are seconds, but you can also qualify numbers with h for hours, m for minutes, d for days, or w for weeks. For example, the lines
$TTL 86400
$TTL 24h
$TTL 1d

all set the $TTL to one day.
Resource records

Each zone of the DNS hierarchy has a set of resource records associated with it. The basic format of a resource record is

[name] [ttl] [class] type data

Fields are separated by whitespace (tabs or spaces) and can contain the special characters shown in Table 16.2.

**Table 16.2: Special characters in resource records**

<table>
<thead>
<tr>
<th>Character</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>;</td>
<td>Introduces a comment</td>
</tr>
<tr>
<td>@</td>
<td>The current zone name</td>
</tr>
<tr>
<td>( )</td>
<td>Allows data to span lines</td>
</tr>
<tr>
<td>*</td>
<td>Wild card (name field only)</td>
</tr>
</tbody>
</table>

The `name` field identifies the entity (usually a host or domain) that the record describes. If several consecutive records refer to the same entity, the name can be omitted after the first record as long as the subsequent records begin with white-space. If present, the `name` field must begin in column one.

A name can be either relative or absolute. Absolute names end with a dot and are complete. Internally, the software deals only with absolute names; it appends the current origin and a dot to any name that does not already end in a dot. This feature allows names to be shorter, but it also invites mistakes.

For example, if cs.colorado.edu were the current domain, the name “anchor” would be interpreted as “anchor.cs.colorado.edu.” If by mistake you entered the name as “anchor.cs.colorado.edu”, the lack of a final dot would still imply a relative name, resulting in the name “anchor.cs.colorado.edu.cs.colorado.edu.” This kind of mistake is common.

The `ttl` (time to live) field specifies the length of time, in seconds, that the record can be cached and still be considered valid. It is often omitted, except in the root server hints file. It defaults to the value set by the `$TTL` directive, which must be the first line of the zone data file.

Increasing the value of the `ttl` parameter to about a week substantially reduces network traffic and DNS load. However, once records have been cached outside your local network, you cannot force them to be discarded. If you plan a massive renumbering and your old `ttl` was a week, lower the `$TTL` value (e.g., to one hour) at least a week before your intended renumbering. This preparatory step makes sure that records with week-long `tts` are expired and replaced with
records that have one-hour \textit{ttl}s. You can then be certain that all your updates will propagate together within an hour. Set the \textit{ttl}s back to their original value after you’ve completed your update campaign.

Some sites set the TTL on the records for Internet-facing servers to a low value so that if a server experiences problems (network failure, hardware failure, denial-of-service attack, etc.), the administrators can respond by changing the server’s name-to-IP-address mapping. Because the original TTLs were low, the new values will propagate quickly. For example, the name google.com has a five-minute TTL, but Google’s name servers have a TTL of four days (345,600 seconds):

\begin{verbatim}
  google.com. 300 IN A 216.58.217.46
  google.com. 345600 IN NS ns1.google.com.
  ns1.google.com. 345600 IN A 216.239.32.10
\end{verbatim}

We used \texttt{dig} to obtain these records; we truncated the output.

The \textit{class} specifies the network type. IN for Internet is the default.

Many different types of DNS records are defined, but fewer than 10 are in common use; IPv6 adds a few more. We divide the resource records into four groups:

- Zone infrastructure records, which identify domains and their name servers
- Basic records, which map between names and addresses and route mail
- Security records, which add authentication and signatures to zone files
- Optional records, which provide extra information about hosts or domains

The contents of the \textit{data} field depend on the record type. A DNS query for a particular domain and record type returns all matching resource records from the zone file. \textit{Table 16.3} lists the common record types.

\textbf{Table 16.3: DNS record types}
Some record types are obsolete, experimental, or not widely used. See your name server’s implementation documentation for a complete list. Most records are maintained by hand (by editing text files or by entering them in a web GUI), but the security resource records require cryptographic processing and so must be managed with software tools. These records are described in the DNSSEC section beginning here.

The order of resource records in the zone file is arbitrary, but traditionally the SOA record is first, followed by the NS records. The records for each host are usually kept together. It’s common practice to sort by the name field, although some sites sort by IP address so that it’s easier to identify unused addresses.

As we describe each type of resource record in detail in the next sections, we inspect some sample records from the atrust.com domain’s data files. The default domain in this context is “atrust.com.”, so a host specified as “bark” really means “bark.atrust.com.”.

See this page for more information about RFCs.

The format and interpretation of each type of resource record is specified by the IETF in the RFC series. In the upcoming sections, we list the specific RFCs relevant to each record type (along with their years of origin) in a margin note.
The SOA record

SOA records are specified in RFC1035 (1987).

An SOA (Start of Authority) record marks the beginning of a zone, a group of resource records located at the same place within the DNS namespace. The data for a DNS domain usually includes at least two zones: one for translating hostnames to IP addresses, called the forward zone, and others that map IP addresses back to hostnames, called reverse zones.

Each zone has exactly one SOA record. The SOA record includes the name of the zone, the primary name server for the zone, a technical contact, and various timeout values. Comments are introduced by a semicolon. Here’s an example:

```
; Start of authority record for atrust.com
atrust.com. IN SOA ns1.atrust.com. hostmaster.atrust.com. (201710200 ; Serial number
10800 ; Refresh (3 hours)
1200 ; Retry (20 minutes)
3600000 ; Expire (40+ days)
3600 ) ; Minimum (1 hour)
```

The *name* field of the SOA record (atrust.com. in this example) often contains the symbol @, which is shorthand for the name of the current zone. The value of @ is the domain name specified in the *zone* statement of named.conf. This value can be changed from within the zone file with the $ORIGIN parser directive (see this page).

This example has no *ttl* field. The class is IN for Internet, the type is SOA, and the remaining items form the *data* field. The numerical parameters in parentheses are timeout values and are often written on one line without comments.

“ns1.atrust.com.” is the zone’s master name server. Actually, any name server for the zone can be listed in the SOA record unless you are using dynamic DNS. In that case, the SOA record must name the master server.

“hostmaster.atrust.com.” was originally intended to be the email address of the technical contact in the format “user.host.” rather than the standard user@host. Unfortunately, due to spam concerns and other reasons, most sites do not keep this contact info updated.

The parentheses continue the SOA record over several lines.

The first numeric parameter is the serial number of the zone’s configuration data. The serial number is used by slave servers to determine when to get fresh data. It can be any 32-bit integer and should be incremented every time the data file for the zone is changed. Many sites encode
the file’s modification date in the serial number. For example, 2017110200 would be the first change to the zone on November 2, 2017.

Serial numbers need not be continuous, but they must increase monotonically. If by accident you set a really large value on the master server and that value is transferred to the slaves, then correcting the serial number on the master will not work. The slaves request new data only if the master’s serial number is larger than theirs.

You can fix this problem in two ways:

- One fix is to exploit the properties of the sequence space in which the serial numbers live. This procedure involves adding a large value ($2^{31}$) to the bloated serial number, letting all the slave servers transfer the data, and then setting the serial number to just what you want. This weird arithmetic, with explicit examples, is covered in detail in the O’Reilly book titled *DNS and BIND*; RFC1982 describes the sequence space.

- A sneaky but more tedious way to fix the problem is to change the serial number on the master, kill the slave servers, remove the slaves’ backup data files so they are forced to reload from the master, and restart the slaves. It does not work to just remove the files and reload; you must kill and restart the slave servers. This method gets hard if you follow best-practices advice and have your slave servers geographically distributed, especially if you are not the sysadmin for those slave servers.

It’s a common mistake to change the data files but forget to update the serial number. Your name server will punish you by failing to propagate your changes to slave servers.

The next four entries in the SOA record are timeout values, in seconds, that control how long data can be cached at various points throughout the world-wide DNS database. Times can also be expressed in units of minutes, hours, days, or weeks by addition of a suffix of m, h, d, or w, respectively. For example, 1h30m means 1 hour and 30 minutes. Timeout values represent a tradeoff between efficiency (it’s cheaper to use an old value than to fetch a new one) and accuracy (new values are more accurate). The four timeout fields are called refresh, update, expire, and minimum.

The refresh timeout specifies how often slave servers should check with the master to see if the serial number of the zone’s configuration has changed. Whenever the zone changes, slaves must update their copy of the zone’s data. The slave compares the serial numbers; if the master’s serial number is larger, the slave requests a zone transfer to update the data. Common values for the refresh timeout range from one to six hours (3,600 to 21,600 seconds).

Instead of just waiting passively for slave servers to time out, master servers for BIND notify their slaves every time a zone changes. However, it’s possible for an update notification to be lost because of network congestion, so the refresh timeout should still be set to a reasonable value.
If a slave server tries to check the master’s serial number but the master does not respond, the slave tries again after the *retry* timeout period has elapsed. Our experience suggests that 20–60 minutes (1,200–3,600 seconds) is a good value.

If a master server is down for a long time, slaves will try to refresh their data many times but always fail. Each slave should eventually decide that the master is never coming back and that its data is surely out of date. The *expire* parameter determines how long the slaves will continue to serve the domain’s data authoritatively in the absence of a master. The system should be able to survive if the master server is down for a few days, so this parameter should have a longish value. We recommend a month or two.

The *minimum* parameter in the SOA record sets the time to live for negative answers that are cached. The default for positive answers (i.e., actual records) is specified at the top of the zone file with the $TTL directive. Experience suggests values of several hours to a few days for $TTL and an hour to a few hours for the *minimum*. BIND silently discards any *minimum* values greater than 3 hours.

The $TTL, *expire*, and *minimum* parameters eventually force everyone that uses DNS to discard old data values. The initial design of DNS relied on the fact that host data was relatively stable and did not change often. However, DHCP, mobile hosts, and the Internet explosion have changed the rules. Name servers are desperately trying to cope with the dynamic update and incremental zone transfer mechanisms described later.
NS records

NS records are specified in RFC1035 (1987).

NS (name server) records identify the servers that are authoritative for a zone (that is, all the master and slave servers) and delegate subdomains to other organizations. NS records are usually placed directly after a zone’s SOA record.

The format is

\[ \text{zone} \ [\text{ttl}] \ [\text{IN}] \ \text{NS} \ \text{hostname} \]

For example:

```
NS ns1.atrust.com.
NS ns2.atrust.com.
booklab NS ubuntu.booklab.atrust.com.
NS ns1.atrust.com.
```

The first two lines define name servers for the atrust.com domain. No \textit{name} is listed because it is the same as the \textit{name} field of the SOA record that precedes the records; the \textit{name} can therefore be left blank. The \textit{class} is also not listed because IN is the default and does not need to be stated explicitly.

The third and fourth lines delegate a subdomain called booklab.atrust.com to the name servers ubuntu.booklab.atrust.com and ns1.atrust.com. These records are actually part of the booklab subdomain, but they must also appear in the parent zone, atrust.com, in order for the delegation to work. In a similar fashion, NS records for atrust.com are stored in the .com zone file to define the atrust.com subdomain and identify its servers. The .com servers refer queries about hosts in atrust.com to the servers listed in NS records for atrust.com within the .com domain.

See this page for more information about delegation.

The list of name servers in the parent zone should be kept up to date with those in the zone itself, if possible. Nonexistent servers listed in the parent zone can delay name service, although clients will eventually stumble onto one of the functioning name servers. If none of the name servers listed in the parent exist in the child, a so-called lame delegation results; see this page.

Extra servers in the child are OK as long as at least one of the child’s servers still has an NS record in the parent. Check your delegations occasionally with \texttt{dig} or \texttt{drill} to be sure they specify an appropriate set of servers; see this page.
A records

A records are specified in RFC1035 (1987).

A (address) records are the heart of the DNS database. They provide the mapping from hostnames to IP addresses. A host usually has one A record for each of its network interfaces. The format is

```
hostname [ttl] [IN] A ipaddr
```

For example:

```
ns1 IN A 63.173.189.1
```

In this example, the name field is not dot-terminated, so the name server adds the default domain to it to form the fully qualified name “ns1.atrust.com.”. The record associates that name with the IP address 63.173.189.1.
AAA records

AAAA records are specified in RFC3596 (2003).

AAAA records are the IPv6 equivalent of A records. Records are independent of the transport protocol used to deliver them; publishing IPv6 records in your DNS zones does not mean that you must answer DNS queries over IPv6.

The format of an AAAA record is

```
hostname [ttl] [IN] AAAA ipaddr
```

For example:

```
f.root-servers.net. IN AAAA 2001:500:2f::f
```

Each colon-separated chunk of the address represents four hex digits, with leading zeros usually omitted. Two adjacent colons stand for “enough zeros to fill out the 128 bits of a complete IPv6 address.” An address can contain at most one such double colon.
PTR records

---

PTR records are specified in RFC1035 (1987).

---

PTR (pointer) records map from IP addresses back to hostnames. As described in The DNS namespace, reverse mapping records live under the in-addr.arpa domain and are named with the bytes of the IP address in reverse order. For example, the zone for the 189 subnet in this example is 189.173.63.in-addr.arpa.

The general format of a PTR record is

```
addr [ttl] [IN] PTR hostname
```

For example, the PTR record in the 189.173.63.in-addr.arpa zone that corresponds to ns1’s A record above is

```
1 IN PTR ns1.atrust.com.
```

The name 1 does not end in a dot and therefore is relative. But relative to what? Not atrust.com—for this sample record to be accurate, the default zone has to be “189.173.63.in-addr.arpa.”.

You can set the zone by putting the PTR records for each subnet in their own file. The default domain associated with the file is set in the name server configuration file. Another way to do reverse mappings is to include records such as

```
1.189 IN PTR ns1.atrust.com.
```

with a default domain of 173.63.in-addr.arpa. Some sites put all reverse records in the same file and use $ORIGIN directives (see this page) to specify the subnet. Note that the hostname ns1.atrust.com ends with a dot to prevent the default domain, 173.63.in-addr.arpa, from being appended to its name.

Since atrust.com and 189.173.63.in-addr.arpa are different regions of the DNS namespace, they constitute two separate zones. Each zone must have its own SOA record and resource records. In addition to defining an in-addr.arpa zone for each real network, you should also define one that takes care of the loopback network (127.0.0.0), at least if you run BIND. See this page for an example.

---

See this page for more details about subnetting.

---

This all works fine if subnets are defined on byte boundaries. But how do you handle the reverse mappings for a subnet such as 63.173.189.0/26, where that last byte can be in any of four
subnets: 0-63, 64-127, 128-191, or 192-255? An elegant hack defined in RFC2317 exploits CNAME resource records to accomplish this feat.

It’s important that A records match their corresponding PTR records. Mismatched and missing PTR records cause authentication failures that can slow your system to a crawl. This problem is annoying in itself; it can also facilitate denial-of-service attacks against any application that requires the reverse mapping to match the A record.

For IPv6, the reverse mapping information that corresponds to an AAAA address record is a PTR record in the ip6.arpa top-level domain.

The “nibble” format reverses an AAAA address record by expanding each colon-separated address chunk to the full 4 hex digits and then reversing the order of those digits and tacking on ip6.arpa at the end. For example, the PTR record that corresponds to our sample AAAA record on this page would be

```
f.0.0.0.0.0.0.0.0.0.0.0.0.0.0.0.0.0.0.0.0.f.2.0.0.0.0.5.0.1.0.0.2.ip6.arpa.
  PTR f.root-servers.net.
```

This line has been folded to fit the page. It’s unfortunately not very friendly for a sysadmin to have to type or debug or even read. Of course, in your actual DNS zone files, the $ORIGIN statement could hide some of the complexity.
MX records

MX records are specified in RFC1035 (1987).

The mail system uses mail exchanger (MX) records to route mail more efficiently. An MX record preempts the destination specified by the sender of a message, in most cases directing the message to a hub at the recipient’s site. This feature puts the flow of mail into a site under the control of local sysadmins instead of senders.

The format of an MX record is

\[ \text{name [ttl] [IN] MX preference host ...} \]

The records below route mail addressed to user@somehost.atrust.com to the machine mailserver.atrust.com if it is up and accepting email. If mailserver is not available, mail goes to mail-relay3.atrust.com. If neither machine named in the MX records is accepting mail, the fallback behavior is to deliver the mail as originally addressed.

```
somehost   IN MX      10 mailserver.atrust.com.
           IN MX      20 mail-relay3.atrust.com.
```

Hosts with low preference values are tried first: 0 is the most desirable, and 65,535 is as bad as it gets.

MX records are useful in many situations:

- When you have a central mail hub or service provider for incoming mail
- When you want to filter mail for spam or viruses before delivering it
- When the destination host is down
- When the destination host isn’t directly reachable from the Internet
- When the local sysadmin knows where mail should be sent better than your correspondents do (i.e., always)

A machine that accepts email on behalf of another host may need to configure its mail transport program to enable this function. See [here](#) and [here](#) for a discussion of how to set up this configuration on sendmail and Postfix email servers, respectively.

Wild card MX records are also sometimes seen in the DNS database:

```
*        IN MX      10 mailserver.atrust.com.
```
At first glance, this record seems like it would save lots of typing and add a default MX record for all hosts. But wild card records don’t quite work as you might expect. They match anything in the *name* field of a resource record that is *not* already listed as an explicit name in another resource record.

Thus, you *cannot* use a star to set a default value for all your hosts. But perversely, you can use it to set a default value for names that are not your hosts. This setup causes lots of mail to be sent to your hub only to be rejected because the hostname matching the star does not in fact belong to your domain. Ergo, avoid wild card MX records.
CNAME records

CNAME records are specified in RFC1035 (1987).

CNAME records assign additional names to a host. These nicknames are commonly used either to associate a function with a host or to shorten a long hostname. The real name is sometimes called the canonical name (hence, “CNAME”). Some examples:

<table>
<thead>
<tr>
<th></th>
<th></th>
<th>IN</th>
<th>CNAME</th>
</tr>
</thead>
<tbody>
<tr>
<td>ftp</td>
<td></td>
<td></td>
<td>anchor</td>
</tr>
<tr>
<td>kb</td>
<td></td>
<td></td>
<td>kibblesnbits</td>
</tr>
</tbody>
</table>

The format of a CNAME record is

nickname [ttl] [IN] CNAME hostname

When DNS software encounters a CNAME record, it stops its query for the nickname and queries for the real name. If a host has a CNAME record, other records (A, MX, NS, etc.) for that host must refer to its real name, not its nickname. (This rule for CNAMEs was explicitly relaxed for DNSSEC, which adds digital signatures to each DNS resource record set. The RRSIG record for the CNAME refers to the nickname.)

CNAME records can nest eight deep. That is, a CNAME record can point to another CNAME, and that CNAME can point to a third CNAME, and so on, up to seven times; the eighth target must be the real hostname. If you use CNAMEs, the PTR record should point to the real name, not a nickname.

You can avoid CNAMEs altogether by publishing A records for both a host’s real name and its nicknames. This configuration makes lookups slightly faster because the extra layer of indirection is not needed.

RFC1033 requires the “apex” of a zone (sometimes called the “root domain” or “naked domain”) to resolve to one or more A (and/or AAAA) records. The use of a CNAME record is forbidden. In other words, you can do this:


but not this:


This restriction is potentially vexatious, especially when you want the apex to point somewhere within a cloud provider’s network and the server’s IP address is subject to change. In this situation, a static A record is not a reliable option.
To fix the problem, you’ll need to use a managed DNS provider (such as AWS Route 53 or CloudFlare) that has developed some kind of system for hacking around the RFC1033 requirement. Typically, these systems let you specify your apex records in a manner similar to a CNAME, but they actually serve A records to the outside world. The DNS provider does the work of keeping the A records synchronized to the actual target.
SRV records

SRV records are specified in RFC2782 (2000).

An SRV record specifies the location of services within a domain. For example, an SRV record lets you query a remote domain for the name of its FTP server. Before SRV, you had to hope the remote sysadmins had followed the prevailing custom and added a CNAME for “ftp” to their server’s DNS records.

SRV records make more sense than CNAMEs for this application and are certainly a better way for sysadmins to move services around and control their use. However, SRV records must be explicitly sought and parsed by clients, so they are not used in all the places they probably should be. They are used extensively by Windows, however.

SRV records resemble generalized MX records with fields that let the local DNS administrator steer and load-balance connections from the outside world. The format is

```
service.proto.name [ttl] [IN] SRV pri weight port target
```

where service is a service defined in the IANA assigned numbers database (see the list at iana.org/numbers.htm), proto is either tcp or udp, name is the domain to which the SRV record refers, pri is an MX-style priority, weight is a weight used for load balancing among several servers, port is the port on which the service runs, and target is the hostname of the server that provides the service. To avoid a second round trip, DNS servers usually return the A record of the target with the answer to a SRV query.

A value of 0 for the weight parameter means that no special load balancing should be done. A value of “.” for the target means that the service is not run at this site.

Here is an example snitched from RFC2782 and adapted for atrust.com:
This example illustrates the use of both the weight parameter (for SSH) and the priority parameter (HTTP). Both SSH servers are used, with the work being split between them. The backup HTTP server is only used when the principal server is unavailable. All other services are blocked, both for TCP and UDP. However, the fact that other services do not appear in DNS does not mean that they are not actually running, just that you can’t locate them through DNS.
TXT records

TXT records are specified in RFC1035 (1987).

A TXT record adds arbitrary text to a host’s DNS records. For example, some sites have a TXT record that identifies them:

```
IN TXT "Applied Trust Engineering, Boulder, CO, USA"
```

This record directly follows the SOA and NS records for the atrust.com zone and so inherits the name field from them.

The format of a TXT record is

```
name [ttl] [IN] TXT info ...
```

All info items must be quoted. Be sure the quotes are balanced—missing quotes wreak havoc with your DNS data because all the records between the missing quote and the next occurrence of a quote mysteriously disappear.

As with other resource records, servers return TXT records in random order. To encode long items such as addresses, use long text lines rather than a collection of several TXT records.

Because TXT records have no particular format, they are sometimes used to add information for other purposes without requiring changes to the DNS system itself.
SPF, DKIM, and DMARC records

SPF (Sender Policy Framework), DKIM (DomainKeys Identified Mail), and DMARC (Domain-based Message Authentication, Reporting, and Conformance) are standards that attempt to stem the Internet’s ever-increasing flow of unsolicited commercial email (aka UCE or spam). Each of these systems distributes spam-fighting information through DNS in the form of TXT records, so they are not true DNS record types. For that reason, we cover these systems in Chapter 18, *Electronic Mail*. See the material that starts here. (There is in fact a defined DNS record type for SPF; however, the TXT record version is preferred.)
DNSSEC records

Five resource record types are currently associated with DNSSEC, the cryptographically secured version of DNS.

DS and DNSKEY records store various types of keys and fingerprints. RRSIGs contain the signatures of other records in the zone (record sets, really). Finally, NSEC and NSEC3 records give DNS servers a way to sign nonexistent records, thus extending cryptographic security to negative query responses. These six records differ from most in that they are generated by software tools rather than being typed in by hand.

DNSSEC is a big topic in its own right, so we discuss these records and their use in the DNSSEC section, which begins [here](#).
16.6 The BIND software

BIND, the Berkeley Internet Name Domain system, is an open source software package from the Internet Systems Consortium (ISC) that implements DNS for Linux, UNIX, macOS, and Windows systems. There have been three main flavors of BIND: BIND 4, BIND 8, and BIND 9, with BIND 10 currently under development by ISC. We cover only BIND 9 in this book.
Components of BIND

The BIND distribution has four major components:

- A name server daemon called **named** that answers queries
- A resolver library that queries DNS servers on behalf of users
- Command-line interfaces to DNS: **nslookup**, **dig**, and **host**
- A program to remotely control **named** called **rndc**

The hardest BIND-related sysadmin chore is probably sorting through all the myriad options and features that BIND supports and determining which ones make sense for your situation.
Configuration files

A complete configuration for named consists of the config file (named.conf), the zone data files that contain address mappings for each host, and the root name server hints file. Authoritative servers need named.conf and zone data files for each zone for which they are the master server; caching servers need named.conf and the root hints file.

named.conf has its own format; all the other files are collections of individual DNS data records whose formats were discussed in The DNS database.

The named.conf file specifies the roles of this host (master, slave, stub, or caching-only) and the manner in which it should obtain its copy of the data for each zone it serves. It’s also the place where options are specified—both global options related to the overall operation of named and server- or zone-specific options that apply to only a portion of the DNS traffic.

The config file consists of a series of statements whose syntax we describe as they are introduced in subsequent sections. The format is unfortunately quite fragile—a missing semicolon or unbalanced quote can wreak havoc.

Comments can appear anywhere that whitespace is appropriate. C, C++, and shell-style comments are all understood:

/* This is a comment that can span lines. */
// Everything to the end of the line is a comment.
# Everything to the end of the line is a comment.

Each statement begins with a keyword that identifies the type of statement. There can be more than one instance of each type of statement, except for options and logging. Statements and parts of statements can also be left out, invoking default behavior for the missing items.

Table 16.4 lists the available statements; the Link column points to our discussion of each statement in the upcoming sections.

Table 16.4: Statements used in named.conf
Before describing these statements and the way they are used to configure `named`, we need to describe a data structure that is used in many of the statements, the address match list. An address match list is a generalization of an IP address that can include the following items:

- An IP address, either IPv4 or IPv6 (e.g., 199.165.145.4 or fe80::202:b3ff:fe1e:8329)
- An IP network specified with a CIDR netmask (e.g., 199.165/16; see this page)
- The name of a previously defined access control list (see this page)
- The name of a cryptographic authentication key
- The `!` character to negate things

Address match lists are used as parameters to many statements and options. Some examples:

```plaintext
{ !1.2.3.13; 1.2.3/24; }
{ 128.138/16; 198.11.16/24; 204.228.69/24; 127.0.0.1; }
```

The first of these lists excludes the host 1.2.3.13 but includes the rest of the 1.2.3.0/24 network; the second defines the networks assigned to the University of Colorado. The braces and final semicolons are not really part of the address match lists but are included here for illustration; they would be part of the enclosing statements of which the address match lists are a part.

When an IP address or network is compared to a match list, the list is searched in order until a match is found. This “first match” algorithm makes the ordering of entries important. For example, the first address match list above would not have the desired effect if the two entries were reversed, because 1.2.3.13 would succeed in matching 1.2.3.0/24 and the negated entry would never be encountered.
Now, on to the statements! Some are short and sweet; others almost warrant a chapter unto themselves.
The include statement

To break up or better organize a large configuration, you can put different portions of the configuration in separate files. Subsidiary files are brought into named.conf with an include statement:

```
include "path";
```

If the path is relative, it is interpreted with respect to the directory specified in the directory option.

A common use of include is to incorporate cryptographic keys that should not be world-readable. Rather than forbidding read access to the entire named.conf file, some sites keep keys in files with restrictive permissions that only named can read. Those files are then included into named.conf.

Many sites put zone statements in a separate file and use the include statement to pull them in. This configuration helps separate the parts of the configuration that are relatively static from those that are likely to change frequently.
The options statement

The options statement specifies global options, some of which might later be overridden for particular zones or servers. The general format is

```
options {
    option;
    option;
    ...
};
```

If no options statement is present in named.conf, default values are used.

BIND has a lot of options—too many, in fact. The 9.9 release has more than 170, which is a lot for sysadmins to wrap their heads around. Unfortunately, as soon as the BIND folks think about removing some of the options that were a bad idea or that are no longer necessary, they get pushback from sites who use and need those obscure options. We do not cover the whole gamut of BIND options here; we have biased our coverage and discuss only the ones whose use we recommend. (We also asked the BIND developers for their suggestions on which options to cover, and we followed their advice.)

For more complete coverage of the options, see one of the books on DNS and BIND listed at the end of this chapter. You can also refer to the documentation shipped with BIND. The ARM document in the doc directory of the distribution describes each option and shows both syntax and default values. The file doc/misc/options also contains a complete list of options.

The default values are listed in square brackets beside each option. For most sites, the default values are just fine. Options are listed in no particular order.

File location options

```
directory "path";  [directory where the server was started]
key-directory "path";  [same as directory entry]
```

The directory statement causes named to cd to the specified directory. Wherever relative pathnames appear in named’s configuration files, they are interpreted relative to this directory. The path should be an absolute path. Any output files (debugging, statistics, etc.) are also written in this directory. The key-directory is where cryptographic keys are stored; it should not be world-readable.

We like to put all the BIND-related configuration files (other than named.conf and resolv.conf) in a subdirectory beneath /var (or wherever you keep your configuration files for other programs). We use /var/named or /var/domain.

Name server identity options
The version string identifies the version of the name server software running on the server. The hostname string identifies the server itself, as does the server-id string. These options let you lie about the true values. Each of them puts data into CHAOS-class (as opposed to IN-class, the default) TXT records where curious onlookers can search for them with the dig command.

The hostname and server-id parameters are additions motivated by the use of anycast routing to duplicate instances of the root and gTLD servers.

**Zone synchronization options**

```plaintext
notify yes | master-only | explicit | no;   [yes]
also-notify server-ipaddr;   [empty]
allow-notify address-match-list;   [empty]
```

The notify and also-notify clauses apply only to master servers. allow-notify applies only to slave servers.

Early versions of BIND synchronized zone files between master and slave servers only when the refresh timeout in the zone’s SOA record had expired. These days, the master named automatically notifies its peers whenever the corresponding zone database has been reloaded, as long as notify is set to yes. The slave servers can then rendezvous with the master to see if the file has changed, and if so, to update their copies of the zone data.

You can use notify both as a global option and as a zone-specific option. It makes the zone files converge much more quickly after you make changes. By default, every authoritative server sends updates to every other authoritative server (a system termed “splattercast” by Paul Vixie). Setting notify to master-only curbs this chatter by sending notifications only to slave servers of zones for which this server is the master. If the notify option is set to explicit, then named only notifies the servers listed in the also-notify clause.

See this page for more information about stub zones.

named normally figures out which machines are slave servers of a zone by looking at the zone’s NS records. If also-notify is specified, a set of additional servers that are not advertised with NS records can also be notified. This tweak is sometimes necessary when your site has internal servers.

The target of an also-notify is a list of IP addresses and, optionally, ports. You must use the allow-notify clause in the secondaries’ named.conf files if you want a name server other than the master to notify them.
For servers with multiple network interfaces, additional options specify the IP address and port to use for outgoing notifications.

**Query recursion options**

```
recursion yes | no;          [yes]
allow-recursion { address-match-list }; [all hosts]
```

The `recursion` option specifies whether `named` should process queries recursively on behalf of your users. You can enable this option on an authoritative server of your zones’ data, but that’s frowned upon. The best-practice recommendation is to keep authoritative servers and caching servers separate.

If this name server should be recursive for your clients, set `recursion` to `yes` and include an `allow-recursion` clause so that `named` can distinguish queries that originate at your site from remote queries. `named` will act recursively for the former and nonrecursively for the latter. If your name server answers recursive queries for everyone, it is called an open resolver and can become a reflector for certain kinds of attacks; see RFC5358.

**Cache memory use options**

```
recursive-clients number;   [1000]
max-cache-size number;      [unlimited]
```

If your server is handling an extraordinary amount of traffic, you may need to tweak the `recursive-clients` and `max-cache-size` options. `recursive-clients` controls the number of recursive lookups the server will process simultaneously; each requires about 20KiB of memory. `max-cache-size` limits the amount of memory the server will use for caching answers to queries. If the cache grows too large, `named` deletes records before their TTLs expire to keep memory use under the limit.

**IP port utilization options**

```
use-v4-udp-ports { range begin end; };   [range 1024 65535]
use-v6-udp-ports { range begin end; };   [range 1024 65535]
avoid-v4-udp-ports { port-list };        [empty]
avoid-v6-udp-ports { port-list };        [empty]
query-source v4-address [port] [any]     # CAUTION, don't use port
query-source-v6 v6-address [port] [any]  # CAUTION, don't use port
```

Source ports have become important in the DNS world because of a DNS protocol weakness discovered by Dan Kaminsky, one that allows DNS cache poisoning when name servers use predictable source ports and query IDs. The `use-` and `avoid-` options for UDP ports (together with changes to the `named` software) have mitigated this attack.
Some sysadmins formerly set a specific outgoing port number so they could configure their firewalls to recognize it and accept UDP packets only for that port. However, this configuration is no longer safe in the post-Kaminsky era. Don’t use the query-source address options to specify a fixed outgoing port for DNS queries or you will forfeit the Kaminsky protection that a large range of random ports provides.

The defaults for the use-* ranges are fine, and you shouldn’t need to change them. But be aware of the implications: queries go out from random high-numbered ports, and the answers come back to those same ports. Ergo, your firewall must be prepared to accept UDP packets on random high-numbered ports.

If your firewall blocks certain ports in this range (for example, port 2049 for RPC) then you have a small problem. When your name server sends a query and uses one of the blocked ports as its source, the firewall blocks the answer. The name server eventually stops waiting and sends out the query again. Not fatal, but annoying to the user caught in the crossfire.

To forestall this problem, use the avoid-* options to make BIND stay away from the blocked ports. Any high-numbered UDP ports blocked by your firewall should be included in the list. (Some firewalls are stateful and may be smart enough to recognize the DNS answer as being paired with the corresponding query of a second ago. Such firewalls don’t need help from this option.) If you update your firewall in response to some threatened attack, be sure to update the port list here, too.

The query-source options let you specify the IP address to be used on outgoing queries. For example, you might need to use a specific IP address to get through your firewall or to distinguish between internal and external views.

*Forwarding options*

    forwarders { in_addr; in_addr; ... }; [empty list]
    forward only | first; [first]

Instead of having every name server perform its own external queries, you can designate one or more servers as forwarders. A run-of-the-mill server can look in its cache and in the records for which it is authoritative. If it doesn’t find the answer it’s looking for, it can then send the query on to a forwarder host. That way, the forwarders build up caches that benefit the entire site. The designation is implicit—nothing in the configuration file of the forwarder explicitly says “Hey, you’re a forwarder.”

The forwarders option lists the IP addresses of the servers you want to use as forwarders. They are queried in turn. The use of a forwarder circumvents the normal DNS procedure of starting at a root server and following the chain of referrals. Be careful not to create forwarding loops.

A forward-only server caches answers and queries forwarders, but it never queries anyone else. If the forwarders do not respond, queries fail. A forward-first server prefers to deal with forwarders, but if they do not respond, the forward-first server will complete queries on its own.
Since the forwarders option has no default value, forwarding does not occur unless it has been specifically configured.

You can turn on forwarding either globally or within individual zone statements.

**Permission options**

```
allow-query { address-match-list };               [all hosts]
allow-query-cache { address-match-list };        [all hosts]
allow-transfer { address-match-list };           [all hosts]
allow-update { address-match-list };             [none]
blackhole { address-match-list };                [empty]
```

These options specify which hosts (or networks) can query your name server or its cache, request block transfers of your zone data, or dynamically update your zones. These match lists are a low-rent form of security and are susceptible to IP address spoofing, so there’s some risk in relying on them. It’s probably not a big deal if someone tricks your server into answering a DNS query, but avoid the allow_update and allow_transfer clauses; use cryptographic keys instead.

The blackhole address list identifies servers that you never want to talk to; named does not accept queries from these servers and will never ask them for answers.

**Packet size options**

```
edns-udp-size number;                              [4096]
max-udp-size number;                                [4096]
```

All machines on the Internet must be capable of reassembling a fragmented UDP packet of 512 bytes or fewer. Although this conservative requirement made sense in the 1980s, it is laughably small by modern standards. Modern routers and firewalls can handle much larger packets, but it only takes one bad link in the IP chain to spoil the whole path.

Since DNS by default uses UDP for queries and since DNS responses are often larger than 512 bytes, DNS administrators have to worry about large UDP packets being dropped. If a large reply gets fragmented and your firewall only lets the first fragment through, the receiver gets a truncated answer and retries the query with TCP. TCP is a more expensive protocol than UDP, and busy servers at the root or TLDs don’t need increased TCP traffic because of everybody’s broken firewalls.

The edns-udp-size option sets the reassembly buffer size that the name server advertises through EDNS0, the extended DNS protocol. The max-udp-size option sets the maximum packet size that the server will actually send. Both sizes are in bytes. Reasonable values are in the 512–4,096 byte range.

**DNSSEC control options**
These options configure support for DNSSEC. See the sections starting on this page for a general discussion of DNSSEC and a detailed description of how to set up DNSSEC at your site.

An authoritative server needs the `dnssec-enable` option turned on. A recursive server needs the `dnssec-enable` and `dnssec-validation` options turned on.

`dnssec-enable` and `dnssec-validation` are turned on by default, which has various implications:

- An authoritative server of a signed zone answering a query with the DNSSEC-aware bit turned on answers with the requested resource records and their signatures.
- An authoritative server of a signed zone answering a query with the DNSSEC-aware bit *not* set answers with just the requested resource records, as in the pre-DNSSEC era.
- An authoritative server of an unsigned zone answers queries with just the requested resource records; there are no signatures to include.
- A recursive server sends queries on behalf of users with the DNSSEC-aware bit set.
- A recursive server validates the signatures included with signed replies before returning data to the user.

The `dnssec-must-be-secure` option allows you to specify that you will only accept secure answers from particular domains, or, alternatively, that you don’t care and that insecure answers are OK. For example, you might say yes to the domain `important-stuff.mybank.com` and no to the domain `marketing.mybank.com`.

### Statistics-gathering option

`zone-statistics yes | no` [no]

This option causes `named` to maintain per-zone statistics as well as global statistics. Run `rndc stats` to dump the statistics to a file.

### Performance tuning options
This long list of options can be used to tune named to run well on your hardware. We don’t describe them in detail, but if you are having performance problems, these options may suggest a starting point for your tuning efforts.

Whew, we are finally done with the options. Let’s get on to the rest of the configuration language!
The acl statement

An access control list is just an address match list with a name:

```plaintext
acl acl-name {
    address-match-list
}
```

You can use an `acl-name` anywhere an address match list is called for.

An `acl` must be a top-level statement in `named.conf`, so don’t try sneaking it in amid your other option declarations. Also keep in mind that `named.conf` is read in a single pass, so access control lists must be defined before they are used. Four lists are predefined:

- any – all hosts
- localnets – all hosts on the local network(s)
- localhost – the machine itself
- none – nothing

The `localnets` list includes all of the networks to which the host is directly attached. In other words, it’s a list of the machine’s network addresses modulo their netmasks.
The (TSIG) key statement

The key statement defines a “shared secret” (that is, a password) that authenticates communication between two servers; for example, between the master server and a slave for a zone transfer, or between a server and the rndc process that controls it. Background information about BIND’s support for cryptographic authentication is given in the DNS security issues section starting on this page. Here, we touch briefly on the mechanics of the process.

To build a key record, you specify both the cryptographic algorithm that you want to use and the shared secret, represented as a base-64-encoded string (see page 561 for details):

```plaintext
key key-id {
    algorithm string;
    secret string;
};
```

As with access control lists, the key-id must be defined with a key statement before it is used. To associate the key with a particular server, just include key-id in the keys clause of that server’s server statement. The key is used both to verify requests from that server and to sign the responses to those requests.

The shared secret is sensitive information and should not be kept in a world-readable file. Use an include statement to bring it into the named.conf file.
**The server statement**

`named` can potentially talk to many servers, not all of them running current software and not all of them even nominally sane. The `server` statement tells `named` about the characteristics of its remote peers. The `server` statement can override defaults for a particular server; it’s not required unless you want to configure keys for zone transfers.

```
server ip_addr {
    bogus yes | no; [no]
    provide-ixfr yes | no; [yes]
    request-ixfr yes | no; [yes]
    keys { key-id; key-id; ... }; [none]
    transfer-source ip-address [port]; [closest interface]
    transfer-source-v6 ipv6-address [port]; [closest interface]
};
```

You can use a `server` statement to override the values of global configuration options for individual servers. Just list the options for which you want nondefault behavior. We have not shown all the server-specific options, just the ones we think you might need. See the BIND documentation for a complete list.

If you mark a server as being `bogus`, `named` won’t send any queries its way. This directive should be reserved for servers that are in fact bogus. `bogus` differs from the global option `blackhole` in that it suppresses only outbound queries. By contrast, the `blackhole` option completely eliminates all forms of communication with the listed servers.

A BIND name server acting as master for a dynamically updated zone performs incremental zone transfers if `provide-ixfr` is set to `yes`. Likewise, a server acting as a slave requests incremental zone transfers from the master if `request-ixfr` is set to `yes`. Dynamic DNS is discussed in detail [here](#).

The `keys` clause identifies a key ID that has been previously defined in a `key` statement for use with transaction signatures (see [this page](#)). Any requests sent to the remote server are signed with this key. Requests originating at the remote server are not required to be signed, but if they are, the signature will be verified.

The `transfer-source` clauses give the IPv4 or IPv6 address of the interface (and optionally, the port) that should be used as a source address (port) for zone transfer requests. This clause is only needed when the system has multiple interfaces and the remote server has specified a specific IP address in its `allow-transfer` clause; the addresses must match.
The masters statement

The masters statement lets you name a set of one or more master servers by specifying their IP addresses and cryptographic keys. You can then use this defined name in the masters clause of zone statements instead of repeating the IP addresses and keys.

How can there be more than one master? See this page.

The masters facility is helpful when multiple slave or stub zones get their data from the same remote servers. If the addresses or cryptographic keys of the remote servers change, you can update the masters statement that introduces them rather than changing many different zone statements.

The syntax is

```
masters name { ip_addr [port ip_port] [key key] ; ... } ;
```
The logging statement

**named** is the current holder of the “most configurable logging system on Earth” award. Syslog put the prioritization of log messages into the programmer’s hands and the disposition of those messages into the sysadmin’s hands. But for a given priority, the sysadmin had no way to say, “I care about this message but not about that message.” BIND added categories that classify log messages by type, and channels that broaden the choices for the disposition of messages. Categories are determined by the programmer, and channels by the sysadmin.

Since logging requires quite a bit of explanation and is somewhat tangential, we discuss it in the debugging section beginning [here](#).
The statistics-channels statement

The statistics-channels statement lets you connect to a running `named` with a browser to view statistics as they are accumulated. Since the stats of your name server might be sensitive, you should restrict access to this data to trusted hosts at your own site. The syntax is

```plaintext
statistics-channels {
    inet (ip-addr | *) port port# allow { address-match-list };

    ...
}
```

You can include multiple `inet-port-allow` sequences. The defaults are open, so be careful! The IP address defaults to `any`, the port defaults to port 80 (normal HTTP), and the allow clause defaults to letting anyone connect. To use statistics channels, `named` must have been compiled with `libxml2`. 
The zone statement

Zone statements are the heart of the named.conf file. They tell named about the zones for which it is authoritative and set the options that are appropriate for managing each zone. A zone statement is also used by a caching server to preload the root server hints (that is, the names and addresses of the root servers, which bootstrap the DNS lookup process).

The exact format of a zone statement varies, depending on the role that named is to play with respect to that zone. The possible zone types are master, slave, hint, forward, stub, and delegation-only. We do not describe stub zones (used by BIND only) or delegation-only zones (used to stop the use of wildcard records in top-level zones to advertise a registrar’s services). The following brief sections describe the other zone types.

Many of the global options covered earlier can become part of a zone statement and override the previously defined values. We have not repeated those options here, except to mention certain ones that are frequently used.

Configuring the master server for a zone

Here’s the format you need for a zone of which this named is the master server:

```yaml
zone "domain-name" {
    type master;
    file "path";
};
```

The domain-name in a zone specification must always appear in double quotes.

The zone’s data is kept on disk in a human-readable (and human-editable) file. The filename has no default, so you must provide a file statement when declaring a master zone. A zone file is just a collection of DNS resource records in the formats described starting here.

Other server-specific attributes are also frequently specified within the zone statement. For example:

```yaml
allow-query { address-match-list };      [any]
allow-transfer { address-match-list };   [any]
allow-update { address-match-list };     [none]
zone-statistics yes | no                 [no]
```

The access control options are not required, but it’s a good idea to use them. They accept any kind of address match list, so you can configure security either in terms of IP addresses or in terms of TSIG encryption keys. As usual, encryption keys are safer.

See this page for more information about dynamic updates.
If dynamic updates are used for this zone, the `allow-update` clause must be present with an address match list that limits the hosts from which updates can occur. Dynamic updates apply only to master zones; the `allow-update` clause cannot be used for a slave zone. Be sure that this clause includes just your own machines (e.g., DHCP servers) and not the whole Internet. (You also need ingress filtering at your firewall; see this page. Better yet, use TSIG for authentication.)

The `zone-statistics` option makes `named` keep track of query/response statistics such as the number and percentage of responses that were referrals, that resulted in errors, or that demanded recursion.

With all these zone-specific options (and about 40 more we have not covered!), the configuration is starting to sound complicated. However, a master zone declaration consisting of nothing but a pathname to the zone file is perfectly reasonable. Here is an example, slightly modified, from the BIND documentation:

```plaintext
zone "example.com" {
    type master;
    file "forward/example.com";
    allow-query { any; };
    allow-transfer { my-slaves; };
}
```

Here, `my-slaves` would be an access control list you had previously defined.

**Configuring a slave server for a zone**

The `zone` statement for a slave is similar to that of a master:

```plaintext
zone "domain-name" {
    type slave;
    file "path";
    masters { ip_addr [port ip_port] [key keyname]; ... };
    allow-query { address-match-list }; [any]
};
```

Slave servers normally maintain a complete copy of their zone’s database. The `file` statement specifies a local file in which the replicated database can be stored. Each time the server fetches a new copy of the zone, it saves the data in this file. If the server crashes and reboots, the file can then be reloaded from the local disk without being transferred across the network.

You shouldn’t edit this cache file, since it’s maintained by `named`. However, it can be interesting to inspect if you suspect you have made an error in the master server’s data file. The slave’s disk file shows you how `named` has interpreted the original zone data. In particular, relative names and `$ORIGIN` directives have all been expanded. If you see a name in the data file that looks like one of these

anchor.cs.colorado.edu.cs.colorado.edu.

you can be pretty sure that you forgot a trailing dot somewhere.

The `masters` clause lists the IP addresses of one or more machines from which the zone database can be obtained. It can also contain the name of a list of masters defined by a previous `masters` statement.

We said that only one machine can be the master for a zone, so why is it possible to list more than one address? Two reasons. First, the master machine might have more than one network interface and therefore more than one IP address. It’s possible for one interface to become unreachable (because of network or routing problems) while others are still accessible. Therefore, it’s a good practice to list all the master server’s topologically distinct addresses.

Second, `named` doesn’t care where the zone data comes from. It can pull the database just as easily from a slave server as from the master. You could use this feature to allow a well-connected slave server to serve as a sort of backup master, since the IP addresses are tried in order until a working server is found. In theory, you can also set up a hierarchy of servers, with one master serving several second-level servers, which in turn serve many third-level servers.

Setting up the root server hints

Another form of `zone` statement points `named` toward a file from which it can preload its cache with the names and addresses of the root name servers.

```plaintext
zone "." {
    type hint;
    file "path";
};
```

The “hints” are a set of DNS records that list servers for the root domain. They’re needed to give a recursive, caching instance of `named` a place to start searching for information about other sites’ domains. Without them, `named` would only know about the domains it serves and their subdomains.

When `named` starts, it reloads the hints from one of the root servers. Ergo, you’ll be fine as long as your hints file contains at least one valid, reachable root server. As a fallback, the root server hints are also compiled into `named`.

The hints file is often called `root.cache`. It contains the response you would get if you queried any root server for the name server records in the root domain. In fact, you can generate the hints file in exactly this way with `dig`. For example:

```
$ dig @f.root-servers.net . ns > root.cache
```

Mind the dot. If f.root-servers.net is not responding, you can run the query without specifying a particular server:
The output will be similar; however, you will be obtaining the list of root servers from the cache of a local name server, not from an authoritative source. That should be just fine—even if you have not rebooted or restarted your name server for a year or two, it has been refreshing its root server records periodically as their TTLs expire.

Setting up a forwarding zone

A zone of type forward overrides named’s default query path (ask the root first, then follow referrals, as described on this page) for a particular domain:

```bash
zone "domain-name" {
    type forward;
    forward only | first;
    forwarders { ip_addr; ip_addr; ... ;
};
```

You might use a forward zone if your organization had a strategic working relationship with some other group or company and you wanted to funnel traffic directly to that company’s name servers, bypassing the standard query path.
The controls statement for rndc

The `controls` statement limits the interaction between the running `named` process and `rndc`, the program a sysadmin can use to signal and control it. `rndc` can start and stop `named`, dump its state, put it in debug mode, etc. `rndc` operates over the network, and with improper configuration it might let anyone on the Internet mess with your name server. The syntax is

```
controls {
    inet addr port port allow { address-match-list } keys { key_list };
}
```

`rndc` talks to `named` on port 953 if you don’t specify a different port.

Allowing your name server to be controlled remotely is both handy and dangerous. Strong authentication through a key entry in the `allow` clause is required; keys in the address match list are ignored and must be explicitly stated in the `keys` clause of the `controls` statement.

You can use the `rndc-confgen` command to generate an authentication key for use between `rndc` and `named`. There are essentially two ways to set up use of the key: you can have both `named` and `rndc` consult the same configuration file to learn the key (e.g., `/etc/rndc.key`), or you can include the key in both the `rndc`’s and `named`’s configuration files (`/etc/rndc.conf` for `rndc` and `/etc/named.conf` for `named`). The latter option is more complicated, but it’s necessary when `named` and `rndc` will be running on different computers. `rndc-confgen -a` sets up keys for localhost access.

When no `controls` statement is present, BIND defaults to the loopback address for the address match list and looks for the key in `/etc/rndc.key`. Because strong authentication is mandatory, the `rndc` command cannot control `named` if no key exists. This precaution may seem draconian, but consider: even if `rndc` worked only from 127.0.0.1 and this address was blocked from the outside world at your firewall, you would still be trusting all local users not to tamper with your name server. Any user could `telnet` to the control port and type “stop”—quite an effective denial-of-service attack.

Here is an example of the output (to standard out) from `rndc-confgen` when a 256-bit key is requested. We chose 256 bits because it fits on the page; you would normally choose a longer key and redirect the output to `/etc/rndc.conf`. The comments at the bottom of the output show the lines you need to add to `named.conf` to make `named` and `rndc` play together.
$ ./rndc-confgen -b 256
# Start of rndc.conf
key "rndc-key" {
    algorithm hmac-md5;
    secret "orZuz5amkUnEp52zlHxD6cd5hAC1d0GsnGe1P/dv2IY=";
};

options {
    default-key "rndc-key";
    default-server 127.0.0.1;
    default-port 953;
};
# End of rndc.conf

# Use the following in named.conf, adjusting the allow list as needed:
# key "rndc-key" {
#    algorithm hmac-md5;
#    secret "orZuz5amkUnEp52zlHxD6cd5hAC1d0GsnGe1P/dv2IY=";
# };
#
# controls {
#    inet 127.0.0.1 port 953
#    allow { 127.0.0.1; } keys { "rndc-key"; };
# };
# End of named.conf
16.7 Split DNS and the View Statement

Many sites want the internal view of their network to be different from the view seen from the Internet. For example, you might reveal all of a zone’s hosts to internal users but restrict the external view to a few well-known servers. Or, you might expose the same set of hosts in both views but supply additional (or different) records to internal users. For example, the MX records for mail routing might point to a single mail hub machine from outside the domain but point to individual workstations from the perspective of internal users.

See this page for more information about private address spaces.

A split DNS configuration is especially useful for sites that use RFC1918 private IP addresses on their internal networks. For example, a query for the hostname associated with IP address 10.0.0.1 can never be answered by the global DNS system, but it is meaningful within the context of the local network. Of the queries arriving at the root name servers, 4%–5% are either from an IP address in one of the private address ranges or about one of these addresses. Neither can be answered; both are the result of misconfiguration, either of BIND’s split DNS or of Microsoft’s “domains.”

The view statement packages up a couple of access lists that control which clients see which view, some options that apply to all the zones in the view, and finally, the zones themselves. The syntax is

```
view view-name {
    match-clients { address-match-list } ; [any]
    match-destinations { address-match-list } ; [any]
    match-recursive-only yes | no; [no]
    view-option; ...
    zone-statement; ...
}
```

Views have always had a match-clients clause that filters on queries’ source IP addresses. It typically serves internal and external views of a site’s DNS data. For finer control, you can now also filter on the query destination address and can require recursive queries.

The match-destinations clause looks at the destination address to which a query was sent. It’s useful on multihomed machines (that is, machines with more than one network interface) when you want to serve different DNS data depending on the interface on which the query arrived. The match-recursive-only clause requires queries to be recursive as well as to originate at a permitted client. Iterative queries let you see what is in a site’s cache; this option prevents it.

Views are processed in order, so put the most restrictive views first. Zones in different views can have the same names but take their data from different files. Views are an all-or-nothing
proposition; if you use them, all `zone` statements in your `named` configuration file must appear in
the context of a `view`.

Here is a simple example from the BIND 9 documentation. The two views define the same zone,
but with different data.

```
view "internal" {
    match-clients { our_nets; }; // Only internal networks
    recursion yes;               // Internal clients only
    zone "example.com" {        // Complete view of zone
        type master;
        file "example-internal.db";
    };
}
view "external" {
    match-clients { any; };     // Allow all queries
    recursion no;               // But no recursion
    zone "example.com" {        // Only "public" hosts
        type master;
        file "example-external.db";
    }
};
```

If the order of the views were reversed, no one would ever see the internal view. Internal hosts
would match the `any` value in the `match-clients` clause of the external view before they reached the
internal view.

Our second DNS configuration example starting on [this page](#) provides an additional example of
views.
16.8 BIND CONFIGURATION EXAMPLES

Now that we have explored the wonders of named.conf, let’s look at two complete configuration examples:

- The localhost zone
- A small security company that uses split DNS
The localhost zone

The IPv4 address 127.0.0.1 refers to a host itself and should be mapped to the name “localhost.” Some sites map the address to “localhost.localdomain.” and some do both. The corresponding IPv6 address is ::1.

If you forget to configure the localhost zone, your site may end up querying the root servers for localhost information. The root servers receive so many of these queries that the operators are considering adding a generic mapping between localhost and 127.0.0.1 at the root level. Other unusual names in the popular “bogus TLD” category are lan, home, localdomain, and domain.

The forward mapping for the name localhost can be defined in the forward zone file for the domain (with an appropriate $ORIGIN statement) or in its own file. Each server, even a caching server, is usually the master for its own reverse localhost domain.

Here are the lines in named.conf that configure localhost:

```conf
zone "localhost" {
    type master;
    file "localhost";
    allow-update { none; };
};

zone "0.0.127.in-addr.arpa" {
    type master;
    file "127.0.0";
    allow-update { none; };
};
```

The corresponding forward zone file, localhost, contains the following lines:

```conf
$TTL 30d
@ IN SOA localhost. postmaster.localhost. ( 2015050801 ; Serial 3600 ; Refresh 1800 ; Retry 604800 ; Expiration 3600 ) ; Minimum

NS localhost.
A 127.0.0.1
```

The reverse file, 127.0.0, contains:
The mapping for the localhost address (127.0.0.1) never changes, so the timeouts can be large. Note the serial number, which encodes the date; the file was last changed in 2015. Also note that only the master name server is listed for the localhost domain. The meaning of @ here is “0.0.127.in-addr.arpa.”.
A small security company

Our second example is for a small company that specializes in security consulting. They run BIND 9 on a recent version of Red Hat Enterprise Linux and use views to implement a split DNS system in which internal and external users see different host data. They also use private address space internally; queries about those addresses should never escape to the Internet to clutter up the global DNS system. Here is their `named.conf` file, reformatted and commented a bit:

```plaintext
options {
    directory "/var/domain";
    version "root@ atrust.com";
    allow-transfer { 82.165.230.84; 71.33.249.193; 127.0.0.1; }; 
    listen-on { 192.168.2.10; 192.168.2.1; 127.0.0.1; 192.168.2.12; }; 
};

include "atrust.key"; // Mode 600 file

controls {
    inet 127.0.0.1 allow { 127.0.0.1; } keys { atkey; };
};

view "internal" {
    match-clients { 192.168.0.0/16; 206.168.198.192/28; 172.29.0.0/24; }; 
    recursion yes;
    include "infrastructure.zones"; // Root hints, localhost forw + rev
    zone "atrust.com" { // Internal forward zone
        type master;
        file "internal/atrust.com";
    }
    zone "1.168.192.in-addr.arpa" { // Internal reverse zone
        type master;
        file "internal/192.168.1.rev";
        allow-update { none; };
    }
    ... // Many zones omitted
    include "internal/tmark.zones"; // atrust.net, atrust.org slaves
}; // End of internal view
```
The file `atrust.key` defines the key named `atkey`:

```plaintext
key "atkey" {
    algorithm hmac-md5;
    secret "shared secret key goes here";
}
```

The file `tmark.zones` includes variations on the name `atrust.com`, both in different top-level domains (net, org, us, info, etc.) and with different spellings (applied-trust.com, etc.). The file `infrastructure.zones` contains the root hints and localhost files.

Zones are organized by view (internal or world) and type (master or slave), and the naming convention for zone data files reflects this scheme. This server is recursive for the internal view, which includes all local hosts, including many that use private addressing. The server is not recursive for the external view, which contains only selected hosts at atrust.com and the external zones for which they provide either master or slave DNS service.

Snippets of the files `internal/atrust.com` and `world/atrust.com` are shown below. First, the `internal` file:
You can see from the IP address ranges that this site is using RFC1918 private addresses internally. Note also that instead of assigning nicknames to hosts through CNAMEs, this site has multiple A records that point to the same IP addresses. This approach works fine, but each IP address should have only one PTR record in the reverse zone.

A records were at one time potentially faster to resolve than CNAMEs because they relieved clients of the need to perform a second DNS query to obtain the address of a CNAME’s target. These days, DNS servers are smarter and automatically include an A record for the target in the original query response (if they know it).

Here is the external view of that same domain from the file world/atrust.com:
As in the internal view, nicknames are implemented with A records. Only a few hosts are actually visible in the external view (although that’s not immediately apparent from these truncated excerpts). Machines that appear in both views (for example, ns1) have RFC1918 private addresses internally but publicly registered and assigned addresses externally.

The TTL in these zone files is set to 16 hours (57,600 seconds). For internal zones, the TTL is one day (86,400 seconds).
16.9 **Zone file updating**

To change a domain’s data (e.g., to add or delete a host), you update the zone data files on the master server. You must also increment the serial number in the zone’s SOA record. Finally, you must get your name server software to pick up and distribute your changes.

This final step varies depending on your software. For BIND, just run `rndc reload` to signal `named` to pick up the changes. You can also kill and restart `named`, but if your server is both authoritative for your zone and recursive for your users, this operation discards cached data from other domains.

Updated zone data is propagated to slave servers of BIND masters right away because the `notify` option is on by default. If notifications are not turned on, your slave servers will not pick up the changes until after `refresh` seconds, as set in the zone’s SOA record (typically an hour later).

If you have the `notify` option turned off, you can force BIND slaves to update themselves by running `rndc reload` on each slave. This command makes the slave check with the master, see that the data has changed, and request a zone transfer.

Don’t forget to modify both the forward and reverse zones when you change a hostname or IP address. Forgetting the reverse files leaves sneaky errors: some commands work and some don’t.

Changing a zone’s data but forgetting to change the serial number makes the changes take effect on the master server (after a reload) but not on the slaves.

Do not edit data files on slave servers. These files are maintained by the name server, and sysadmins should not meddle with them. It’s fine to look at the BIND data files as long as you don’t make changes.
Zone transfers

DNS servers are synchronized through a mechanism called a zone transfer. A zone transfer can include the entire zone (called AXFR) or be limited to incremental changes (called IXFR). By default, zone transfers use the TCP protocol on port 53. BIND logs transfer-related information with category “xfer-in” or “xfer-out.”

A slave wanting to refresh its data must request a zone transfer from the master server and make a backup copy of the zone data on disk. If the data on the master has not changed, as determined by a comparison of the serial numbers (not the actual data), no update occurs and the backup files are just touched. (That is, their modification times are set to the current time.)

Both the sending and receiving servers remain available to answer queries during a zone transfer. Only after the transfer is complete does the slave begin to use the new data.

When zones are huge (like com) or dynamically updated (see the next section), changes are typically small relative to the size of the entire zone. With IXFR, only the changes are sent (unless they are larger than the complete zone, in which case a regular AXFR transfer is done). The IXFR mechanism is analogous to the patch program in that it makes changes to an old database to bring it into conformity with a new database.

In BIND, IXFR is the default for any zones configured for dynamic update, and named maintains a transaction log called zonename.jnl. You can set the options provide-ixfr and request-ixfr in the server statements for individual peers. The provide-ixfr option enables or disables IXFR service for zones for which this server is the master. The request-ixfr option requests IXFRs for zones for which this server is a slave.

provide-ixfr yes;  # In BIND server statement
request-ixfr yes;  # In BIND server statement

IXFRs work for zones that are edited by hand, too. Use the BIND zone option called ixfr-from-differences to enable this behavior. IXFR requires the zone file to be sorted in a canonical order. An IXFR request to a server that does not support it automatically falls back to the standard AXFR zone transfer.
DNS was originally designed under the assumption that name-to-address mappings are relatively stable and do not change frequently. However, a site that uses DHCP to dynamically assign IP addresses as machines boot and join the network breaks this rule constantly. Two basic solutions are available: either add generic (and static) entries to the DNS database, or provide some way to make small, frequent changes to zone data.

The first solution should be familiar to anyone who has looked up the PTR record for the IP address assigned to them by a mass-market (home) ISP. The DNS configuration usually looks something like this:

dhcp-host1.domain. IN A 192.168.0.1
dhcp-host2.domain. IN A 192.168.0.2
...

Although this is a simple solution, it means that hostnames are permanently associated with particular IP addresses and that computers therefore change hostnames whenever they receive new IP addresses. Hostname-based logging and security measures become very difficult in this environment.

The dynamic update feature outlined in RFC2136 offers an alternative solution. It extends the DNS protocol to include an update operation, thereby allowing entities such as DHCP daemons to notify name servers of the address assignments they make. Dynamic updates can add, delete, or modify resource records.

When dynamic updates are enabled in BIND, named maintains a journal of dynamic changes (zonename.jnl) that it can consult in the event of a server crash. named recovers the in-memory state of the zone by reading the original zone files and then replaying the changes from the journal.

You cannot hand-edit a dynamically updated zone without first stopping the dynamic update stream. **rndc freeze zone** or **rndc freeze zone class view** will do the trick. These commands sync the journal file to the master zone file on disk and then delete the journal. You can then edit the zone file by hand. Unfortunately, the original formatting of the zone file will have been destroyed by named’s monkeying—the file will look like those maintained by named for slave servers.

Dynamic update attempts are refused while a zone is frozen. To reload the zone file from disk and reenable dynamic updates, use **rndc thaw** with the same arguments you used to freeze the zone.
The `nsupdate` program supplied with BIND 9 comes with a command-line interface for making
dynamic updates. It runs in batch mode, accepting commands from the keyboard or a file. A
blank line or the `send` command signals the end of an update and sends the changes to the server.
Two blank lines signify the end of input. The command language includes a primitive if
statement to express constructs such as “if this hostname does not exist in DNS, add it.” As
predicates for an `nsupdate` action, you can require a name to exist or not exist, or require a
resource record set to exist or not exist.

For example, here is a simple `nsupdate` script that adds a new host and also adds a nickname for
an existing host if the nickname is not already in use. The angle bracket prompt is produced by
`nsupdate` and is not part of the command script.

```
$ nsupdate
> update add newhost.cs.colorado.edu 86400 A 128.138.243.16
> 
> prereq nxdomain gypsy.cs.colorado.edu
> update add gypsy.cs.colorado.edu CNAME evi-laptop.cs.colorado.edu
```

Dynamic updates to DNS are scary. They can potentially provide uncontrolled write access to
your important system data. Don’t try to use IP addresses for access control—they are too easily
forged. TSIG authentication with a shared-secret key is better; it’s available and is easy to
configure. BIND 9 supports both:

```
$ nsupdate -k keydir: keyfile
```

or

```
$ nsupdate -y keyname: secretkey
```

Since the password goes on the command line in the `-y` form, anyone running `w` or `ps` at the right
moment can see it. For this reason, the `-k` form is preferred. For more details on TSIG, see the
section starting [here](#).

Dynamic updates to a zone are enabled in `named.conf` with an `allow-update` or `update-policy` clause.
`allow-update` grants permission to update any records in accordance with IP- or key-based
authentication. `update-policy` is a BIND 9 extension that allows fine-grained control for updates
according to the hostname or record type. It requires key-based authentication. Both forms can be
used only within `zone` statements, and they are mutually exclusive within a particular `zone`.

A good default for zones with dynamic hosts is to use `update-policy` to allow clients to update their
A or PTR records but not to change the SOA record, NS records, or KEY records.

The syntax of an `update-policy` rule (of which there can be several) is

```
(grant|deny) identity nametype name [types];
```
The identity is the name of the cryptographic key needed to authorize the update. The nametype has one of four values: name, subdomain, wildcard, or self. The self option is particularly prized because it allows hosts to update only their own records. Use it if your situation allows.

The name is the zone to be updated, and the types are the resource record types that can be updated. If no types are specified, all types except SOA, NS, RRSIG, and NSEC or NSEC3 can be updated.

Here’s a complete example:

```bash
update-policy { grant dhcp-key subdomain dhcp.cs.colorado.edu A };
```

This configuration allows anyone who knows the key dhcp-key to update address records in the dhcp.cs.colorado.edu subdomain. This statement would appear in the master server’s named.conf file within the zone statement for dhcp.cs.colorado.edu. (There would be a key statement somewhere to define dhcp-key as well.)

The snippet below from the named.conf file at the computer science department at the University of Colorado uses the update-policy statement to allow students in a system administration class to update their own subdomains but not to mess with the rest of the DNS environment.

```bash
zone "saclass.net" {
  type master;
  file "saclass/saclass.net";
  update-policy {
    grant feanor_mroe. subdomain saclass.net.;
    grant mojo_mroe. subdomain saclass.net.;
    grant dawdle_mroe. subdomain saclass.net.;
    grant pirate_mroe. subdomain saclass.net.;
    ...
  };
  ...
}
```
16.10 DNS security issues

DNS started out as an inherently open system, but it has steadily grown more and more secure—or at least, securable. By default, anyone on the Internet can investigate your domain with individual queries from tools such as `dig`, `host`, `nslookup`, and `drill`. In some cases, they can dump your entire DNS database.

To address such vulnerabilities, name servers support various types of access control that key off of host and network addresses or cryptographic authentication. Table 16.5 summarizes the security features that can be configured in `named.conf`. The Link column points to more information for each topic.

### Table 16.5: Security features in BIND

<table>
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<tr>
<th>Feature</th>
<th>Context</th>
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<tr>
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</tr>
<tr>
<td>update-policy</td>
<td>zone</td>
<td>here</td>
<td>Who can make dynamic updates</td>
</tr>
</tbody>
</table>

BIND can run in a chrooted environment under an unprivileged UID to minimize security risks. It can also use transaction signatures to control communication between master and slave servers and between the name servers and their control programs.
Access control lists in BIND, revisited

ACLs are named address-match lists that can appear as arguments to statements such as `allow-query`, `allow-transfer`, and `blackhole`. Their basic syntax was described [here](#). ACLs can help beef up DNS security in a variety of ways.

Every site should at least have one ACL for bogus addresses and one ACL for local addresses. For example:

```plaintext
acl bogusnets { // ACL for bogus networks
    0.0.0.0/8 ; // Default, wild card addresses
    1.0.0.0/8 ; // Reserved addresses
    2.0.0.0/8 ; // Reserved addresses
    169.254.0.0/16 ; // Link-local delegated addresses
    192.0.2.0/24 ; // Sample addresses, like example.com
    224.0.0.0/3 ; // Multicast address space
    10.0.0.0/8 ; // Private address space (RFC1918)
    172.16.0.0/12 ; // Private address space (RFC1918)
    192.168.0.0/16 ; // Private address space (RFC1918)
};

acl cunets { // ACL for University of Colorado networks
    128.138.0.0/16 ; // Main campus network
    198.11.16/24 ;
    204.228.69/24 ;
};
```

In the global `options` section of your config file, you could then include:

```plaintext
allow-recursion { cunets; };
blackhole { bogusnets; };
```

It’s also a good idea to restrict zone transfers to legitimate slave servers. An ACL makes things nice and tidy:

```plaintext
acl ourslaves {
    128.138.242.1 ; // anchor
    ...
}

acl measurements {
    198.32.4.0/24 ; // Bill manning's measurements, v4 address
    2001:478:6:0::/48 ; // Bill manning's measurements, v6 address
}
```

The actual restriction is implemented with a line such as:

```plaintext
allow-transfer { ourslaves; measurements; };
```
Here, transfers are limited to our own slave servers and to the machines of an Internet measurement project that walks the reverse DNS tree to determine the size of the Internet and the percentage of misconfigured servers. Limiting transfers in this way makes it impossible for other sites to dump your entire database with a tool such as `dig` (see this page).

Of course, you should still protect your network at a lower level through router access control lists and standard security hygiene on each host. If those measures are not possible, you can refuse DNS packets except to a gateway machine that you monitor closely.
Open resolvers

An open resolver is a recursive, caching name server that accepts and answers queries from anyone on the Internet. Open resolvers are bad. Outsiders can consume your resources without your permission or knowledge, and if they are bad guys, they might be able to poison your resolver’s cache.

Worse, open resolvers are sometimes used by miscreants to amplify distributed denial of service attacks. The attacker sends queries to your resolver with a faked source address that points back to the victim of the attack. Your resolver dutifully answers the queries and sends some nice fat packets to the victim. The victim didn’t initiate the queries, but it still has to route and process the network traffic. Multiply by a bunch of open resolvers and it’s real trouble for the victim.

Statistics show that between 70% and 75% of caching name servers are currently open resolvers —yikes! The site dns.measurement-factory.com/tools can help you test your site. Go there, select the “open resolver test,” and type in the IP addresses of your name servers. Alternatively, you can enter a network number or WHOIS identifier to test all the associated servers.

Use access control lists in named.conf to limit your caching name servers to answering queries from your own users.
Running in a chrooted jail

If hackers compromise your name server, they can potentially gain access to the system under the guise of the user as whom it runs. To limit the damage that someone could do in this situation, you can run the server in a chrooted environment, run it as an unprivileged user, or both.

For named, the command-line flag `-t` specifies the directory to chroot to, and the `-u` flag specifies the UID under which named should run. For example,

```
$ sudo named -u 53
```

initially starts named as root, but after named completes its rootly chores, it relinquishes its root privileges and runs as UID 53.

Many sites don’t bother to use the `-u` and `-t` flags, but when a new vulnerability is announced, they must be faster to upgrade than the hackers are to attack.

The chroot jail cannot be empty since it must contain all the files the name server normally needs to run: `/dev/null`, `/dev/random`, the zone files, configuration files, keys, syslog target files, UNIX domain socket for syslog, `/var`, etc. It takes a bit of work to set all this up. The chroot system call is performed after libraries have been loaded, so you need not copy shared libraries into the jail.
Secure server-to-server communication with TSIG and TKEY

During the time when DNSSEC (covered in the next section) was being developed, the IETF developed a simpler mechanism called TSIG (RFC2845) to allow secure communication among servers through the use of “transaction signatures.” Access control through transaction signatures is more secure than access control by IP source addresses alone. TSIG can secure zone transfers between a master server and its slaves and can also secure dynamic updates.

The TSIG seal on a message authenticates the peer and verifies that the data has not been tampered with. Signatures are checked at the time a packet is received and are then discarded; they are not cached and do not become part of the DNS data.

TSIG uses symmetric encryption. That is, the encryption key is the same as the decryption key. This single key is called the “shared secret.” The TSIG specification allows multiple encryption methods, and BIND implements quite a few. Use a different key for each pair of servers that want to communicate securely.

TSIG is much less expensive computationally than public key cryptography, but because it requires manual configuration, it is only appropriate for a local network on which the number of pairs of communicating servers is small. It does not scale to the global Internet.
Setting up TSIG for BIND

First, use BIND’s `dnssec-keygen` utility to generate a shared-secret host key for the two servers, say, master and slave1:

```
$ dnssec-keygen -a HMAC-SHA256 -b 128 -n HOST master-slave1
Kmaster-slave1.+163+15496
```

The `-b 128` flag tells `dnssec-keygen` to create a 128-bit key. We use 128 bits here just to keep the keys short enough to fit on our printed pages. In real life, you might want to use a longer key; 512 bits is the maximum allowed.

This command produces the following two files:

- `Kmaster-slave1.+163+15496.private`
- `Kmaster-slave1.+163+15496.key`

The 163 represents the SHA-256 algorithm, and 15496 is a number used as a key identifier in case you have multiple keys for the same pair of servers. The number looks random, but it is actually just a hash of the TSIG key. Both files include the same key, but in different formats.

The `.private` file looks like this:

```
Private-key-format: v1.3
Algorithm: 163 (HMAC_SHA256)
Key: owKt6ZW0u0gaVFkw0qGxA==
Bits: AAA=
Created: 20160218012956
Publish: 20160218012956
Activate: 20160218012956
```

And the `.key` file like this:

```
master-slave1. IN KEY 512 3 163 owKt6ZW0u0gaVFkw0qGxA==
```

Note that `dnssec-keygen` added a dot to the end of the key names in both the filenames and the contents of the `.key` file. The motivation for this convention is that when `dnssec-keygen` is used for DNSSEC keys that are added to zone files, the key names must be fully qualified domain names and must therefore end in a dot. There should probably be two tools, one that generates shared-secret keys and one that generates public-key key pairs.

You don’t actually need the `.key` file—it’s an artifact of `dnssec-keygen`’s being used for two different jobs. Just delete it. The 512 in the KEY record is not the key length but rather a flag bit that identifies the record as a DNS host key.

After all this complication, you may be disappointed to learn that the generated key is really just a long random number. You could generate the key manually by writing down an ASCII string of
the right length (divisible by 4) and pretending that it’s a base-64 encoding of something, or you could use `mmencode` to encode a random string. The way you create the key is not important; it just has to exist on both machines.

```
scp is part of the OpenSSH suite. See this page for details.
```

Copy the key from the `.private` file to both master and slave1 with `scp`, or cut and paste it. Do not use `telnet` or `ftp` to copy the key; even internal networks might not be secure.

The key must be included in both machines’ `named.conf` files. Since `named.conf` is usually world-readable and keys should not be, put the key in a separate file that is included in `named.conf`. The key file should have mode 600 and should be owned by the `named` user.

For example, you could put the snippet

```plaintext
key master-slave1. {
    algorithm hmac-md5;
    secret "shared-key-you-generated";
};
```

in the file `master-slave1.tsig`. In the `named.conf` file, add the line

```
include "master-slave1.tsig";
```

near the top.

This part of the configuration simply defines the keys. For them to actually be used to sign and verify updates, the master needs to require the key for transfers and the slave needs to identify the master with a `server` statement and `keys` clause. For example, you might add the line

```
allow-transfer { key master-slave1. ;} ;
```

to the `zone` statement on the master server, and the line

```
server master's-IP-address { keys { master-slave1. ; } ; } ;
```

to the slave’s `named.conf` file. If the master server allows dynamic updates, it can also use the key in its `allow-update` clause in the `zone` statement.

Our example key name is pretty generic. If you use TSIG keys for many zones, you might want to include the name of the zone in the key name to help you keep everything straight.

When you first turn on transaction signatures, run `named` at debug level 1 (see this page for information about debug mode) for a while to see any error messages that are generated.
When using TSIG keys and transaction signatures between master and slave servers, keep the clocks of the servers synchronized with NTP. If the clocks are too far apart (more than about 5 minutes), signature verification will not work. This problem can be very hard to identify.

TKEY is a BIND mechanism that lets two hosts generate a shared-secret key automatically, without phone calls or secure copies to distribute the key. It uses an algorithm called the Diffie-Hellman key exchange in which each side makes up a random number, does some math on it, and sends the result to the other side. Each side then mathematically combines its own number with the transmission it received to arrive at the same key. An eavesdropper might overhear the transmission but will be unable to reverse the math. (The math involved is called the discrete log problem and relies on the fact that for modular arithmetic, taking powers is easy but taking logs to undo the powers is close to impossible.)

Microsoft servers use TSIG in a nonstandard way called GSS-TSIG that exchanges the shared secret through TKEY. If you need a Microsoft server to communicate with BIND, use the tkey-domain and tkey-gssapi-credential options.

SIG(0) is another mechanism for signing transactions between servers or between dynamic updaters and the master server. It uses public key cryptography; see RFCs 2535 and 2931 for details.
DNSSEC

DNSSEC is a set of DNS extensions that authenticate the origin of zone data and verify its integrity by using public key cryptography. That is, the extensions allow DNS clients to ask the questions “Did this DNS data really come from the zone’s owner?” and “Is this really the data sent by that owner?”

DNSSEC relies on a cascading chain of trust. The root servers validate information for the top-level domains, the top-level domains validate information for the second-level domains, and so on.

Public key cryptosystems use two keys: one to encrypt (sign) and a different one to decrypt (verify). Publishers sign their data with the secret “private” key. Anyone can verify the validity of a signature with the matching “public” key, which is widely distributed. If a public key correctly decrypts a zone file, then the zone must have been encrypted with the corresponding private key. The trick is to make sure that the public keys you use for verification are authentic. Public key systems allow one entity to sign the public key of another, thereby vouching for the legitimacy of the key; hence the term “chain of trust.”

The data in a DNS zone is too voluminous to be encrypted with public key cryptography—the encryption would be too slow. Instead, since the data is not secret, a secure hash is run on the data and the results of the hash are signed (encrypted) by the zone’s private key. The results of the hash are like a fingerprint of the data and are called a digital signature. The signatures are appended to the data they authenticate as RRSIG records in the signed zone file.

To verify the signature, you decrypt it with the public key of the signer, run the data through the same secure hash algorithm, and compare the computed hash value with the decrypted hash value. If they match, you have authenticated the signer and verified the integrity of the data.

In the DNSSEC system, each zone has its own public and private keys. In fact, it has two sets of keys: a zone-signing key pair and a key-signing key pair. The private zone-signing key signs each RRset (that is, each set of records of the same type for the same host). The public zone-signing key verifies the signatures and is included in the zone’s data in the form of a DNSKEY resource record.

Parent zones contain DS records that are hashes of the child zones’ self-signed key-signing-key DNSKEY records. A name server verifies the authenticity of a child zone’s DNSKEY record by checking it against the parent zone’s signature. To verify the authenticity of the parent zone’s key, the name server can check the parent’s parent, and so on back to the root. The public key for the root zone is widely published and is included in the root hints file.

The DNSSEC specifications require that if a zone has multiple keys, each is tried until the data is validated. This behavior is required so that keys can be rolled over (changed) without interruptions in DNS service. If a DNSSEC-aware recursive name server queries an unsigned zone, the unsigned answer that comes back is accepted as valid. But problems occur when
signatures expire or when parent and child zones do not agree on the child’s current DNSKEY record.
DNSSEC policy

Before you begin deployment of DNSSEC, you should nail down (or at least think about) a few policies and procedures. For example:

- What size keys will you use? Longer keys are more secure, but they make for larger packets.
- How often will you change keys in the absence of a security incident?

We suggest that you keep a key log that records the date you generated each key, the hardware and operating system used, the key tag assigned, the version of the key generator software, the algorithm used, the key length, and the signature validity period. If a cryptographic algorithm is later compromised, you can check your log to see if you are vulnerable.
DNSSEC resource records

DNSSEC uses five resource record types that were referred to in the DNS database section back on this page but were not described in detail: DS, DNSKEY, RRSIG, NSEC, and NSEC3. We describe them here in general and then outline the steps involved in signing a zone. Each of these records is created by DNSSEC tools rather than by being typed into a zone file with a text editor.

The DS (Designated Signer) record appears only in the parent zone and indicates that a subzone is secure (signed). It also identifies the key used by the child to self-sign its own KEY resource record set. The DS record includes a key identifier (a five-digit number), a cryptographic algorithm, a digest type, and a digest of the public key record allowed (or used) to sign the child’s key resource record. Here’s an example.

```
example.com. IN DS 682 5 1 12898DCF9F7C2E89A1AD20DBCE159E7...
```

(In this section, base-64-encoded hashes and keys have all been truncated to save space and better illustrate the structure of the records.)

The question of how to change existing keys in the parent and child zones has been a thorny one that seemed destined to require cooperation and communication between parent and child. The creation of the DS record, the use of separate key-signing and zone-signing keys, and the use of multiple key pairs have helped address this problem.

Keys included in a DNSKEY resource record can be either key-signing keys (KSKs) or zone-signing keys (ZSKs). A flag called SEP for “secure entry point” distinguishes them. Bit 15 of the flags field is set to 1 for KSKs and to 0 for ZSKs. This convention makes the flags field of KSKs odd and of ZSKs even when they are treated as decimal numbers. The values are currently 257 and 256, respectively.

Multiple keys can be generated and signed so that a smooth transition from one key to the next is possible. The child can change its zone-signing keys without notifying the parent; it must only coordinate with the parent if it changes its key-signing key. As keys roll over, both the old key and the new key are valid for a certain interval. Once cached values on the Internet have expired, the old key can be retired.

An RRSIG record is the signature of a resource record set (that is, the set of all records of the same type and name within a zone). RRSIG records are generated by zone-signing software and added to the signed version of the zone file.

An RRSIG record contains a wealth of information:

- The type of record set being signed
- The signature algorithm used, encoded as a small integer
- The number of labels (dot-separated pieces) in the name field
- The TTL of the record set that was signed
- The time the signature expires (as `yyyymmddhhssss`)
- The time the record set was signed (also `yyyymmddhhssss`)
- A key identifier (a 5-digit number)
- The signer’s name (domain name)
- And finally, the digital signature itself (base-64-encoded)

Here’s an example:

```
RRSIG NS 5 2 57600 20090919182841 ( 20090820182841 23301 example.com.
pMKZ76waPVTbIguEQNUojNV1VewHau4p...== )
```

NSEC or NSEC3 records are also produced as a zone is signed. Rather than signing record sets, they certify the intervals between record set names and so allow for a signed answer of “no such domain” or “no such resource record set.” For example, a server might respond to a query for A records named bork.atrust.com with an NSEC record that certifies the nonexistence of any A records between bark.atrust.com and bundt.atrust.com.

Unfortunately, the inclusion of the endpoint names in NSEC records allows someone to walk through the zone and obtain all of its valid hostnames. NSEC3 fixes this feature by including hashes of the endpoint names rather than the endpoint names themselves, but it is more expensive to compute: more security, less performance. NSEC and NSEC3 are both in current use, and you can choose between them when you generate your keys and sign your zones.

Unless protecting against a zone walk is critically important for your site, we recommend that you use NSEC for now.
**Turning on DNSSEC**

Two separate workflows are involved in deploying signed zones: a first that creates keys and signs zones, and a second that serves the contents of those signed zones. These duties need not be implemented on the same machine. In fact, it is better to quarantine the private key and the CPU-intensive signing process on a machine that is not publicly accessible from the Internet. (Of course, the machine that actually serves the data must be visible to the Internet.)

The first step in setting up DNSSEC is to organize your zone files so that all the data files for a zone are in a single directory. The tools that manage DNSSEC zones expect this organization.

Next, enable DNSSEC on your servers with the `named.conf` options

```plaintext
options {
    dsnsec-enable yes;
}
```

for authoritative servers, and

```plaintext
options {
    dsnsec-enable yes;
    dnssec-validation yes;
}
```

for recursive servers. The `dsnsec-enable` option tells your authoritative servers to include DNSSEC record set signatures in their responses when answering queries from DNSSEC-aware name servers. The `dsnsec-validation` option makes `named` verify the legitimacy of signatures it receives in responses from other servers.
Key pair generation

You must generate two key pairs for each zone you want to sign: a zone-signing (ZSK) pair and a key-signing (KSK) pair. Each pair consists of a public key and a private key. The KSK’s private key signs the ZSK and creates a secure entry point for the zone. The ZSK’s private key signs the zone’s resource records. The public keys are then published to allow other sites to verify your signatures.

The commands

```
$ dnssec-keygen -a RSASHA256 -b 1024 -n ZONE example.com
Kexample.com.+008+29718
$ dnssec-keygen -a RSASHA256 -b 2048 -n ZONE -f KSK example.com
Kexample.com.+008+05005
```

generate for example.com a 1,024-bit ZSK pair that uses the RSA and SHA-256 algorithms and a corresponding 2,048-bit KSK pair. The outstanding issue of UDP packet size limits suggests that it’s best to use short zone-signing keys and to change them often. You can use longer key-signing keys to help recover some security.

It can take a while—minutes—to generate these keys. The limiting factor is typically not CPU power but the entropy available for randomization. On Linux, you can install the `haveged` daemon to harvest entropy from additional sources and thereby speed up key generation.

`dnssec-keygen` prints to standard out the base filename of the keys it has generated. In this example, `example.com` is the name of the key, `008` is the identifier of the RSA/SHA-256 algorithm suite, and `29718` and `05005` are hashes called the key identifiers, key footprints, or key tags. As when generating TSIG keys, each run of `dnssec-keygen` creates two files (`.key` and `.private`):

```
Kexample.com.+008+29718.key  # Public zone-signing key
Kexample.com.+008+29718.private# Private zone-signing key
```

Several encryption algorithms are available, each with a range of possible key lengths. You can run `dnssec-keygen` with no arguments to see the current list of supported algorithms. BIND can also use keys generated by other software.

Depending on the version of your software, some of the available algorithm names might have NSEC3 appended or prepended to them. If you want to use NSEC3 records instead of NSEC records for signed negative answers, you must generate NSEC3-compatible keys with one of the NSEC3-specific algorithms; see the man page for `dnssec-keygen`.

The `.key` files each contain a single DNSKEY resource record for example.com. For example, here is the zone-signing public key, truncated to fit the page. You can tell it’s a ZSK because the flags field is 256, rather than 257 for a KSK.
These public keys must be included or inserted into the zone file, either at the end or right after the SOA record. To copy the keys into the zone file, you can append them with `cat` or paste them in with a text editor. Use a command like `cat Kexample.com.+*.key >> zonefile`. The `>>` appends to the `zonefile` rather than replacing it entirely, as `>` would. (Don’t mess this one up!)

Ideally, the private key portion of any key pair would be kept off-line, or at least on a machine that is not on the public Internet. This precaution is impossible for dynamically updated zones and impractical for zone-signing keys, but it is perfectly reasonable for key-signing keys, which are presumably quite long-lived. Consider a hidden master server that is not accessible from outside for the ZSKs. Print out the private KSK or write it to a USB memory stick and then lock it in a safe until you need it again.

While you’re locking away your new private keys, it’s also a good time to enter the new keys into your key log file. You don’t need to include the keys themselves, just the IDs, algorithms, date, purpose, and so on.

The default signature validity periods are one month for RRSIG records (ZSK signatures of resource record sets) and three months for DNSKEY records (KSK signatures of ZSKs). Current best practice suggests ZSKs of length 1,024 that are used for three months to a year and KSKs of length 1,280 that are used for a year or two. The website keylength.com tabulates a variety of organizations’ recommendations regarding the suggested lengths of cryptographic keys.

Since the recommended key retention periods are longer than the default signature validity periods, you must either specify a longer validity period when signing zones or periodically re-sign the zones, even if the key has not changed.
**Zone signing**

Now that you’ve got keys, you can sign your zones with the `dnssec-signzone` command, which adds RRSIG and NSEC or NSEC3 records for each resource record set. These commands read your original zone file and produce a separate, signed copy named `zonefile.signed`.

The syntax is

```
dnssec-signzone [-o zone] [-N increment] [-k KSKfile] zonefile [ZSKfile]
```

where `zone` defaults to `zonefile` and the key files default to the filenames produced by `dnssec-keygen` as outlined above.

If you name your zone data files after the zones and maintain the names of the original key files, the command reduces to

```
dnssec-signzone [-N increment] zonefile
```

The `-N increment` flag automatically increments the serial number in the SOA record so that you can’t forget. You can also specify the value `unixtime` to update the serial number to the current UNIX time (seconds since January 1, 1970) or the value `keep` to prevent `dnssec-signzone` from modifying the original serial number. The serial number is incremented in the signed zone file but not in the original zone file.

Here’s a spelled-out example that uses the keys generated above:

```
$ sudo dnssec-signzone -o example.com -N increment
        -k Kexample.com.+008+05005 example.com Kexample.com.+008+29718
```

The signed file is sorted in alphabetical order and includes the DNSKEY records we added by hand and the RRSIG and NSEC records generated during signing. The zone’s serial number has been incremented.

If you generated your keys with an NSEC3-compatible algorithm, you would sign the zone as above but with a `-3 salt` flag. Table 16.6 shows some other useful options.

**Table 16.6: Useful options for dnssec-signzone**

<table>
<thead>
<tr>
<th>Option</th>
<th>Function</th>
</tr>
</thead>
<tbody>
<tr>
<td><code>-g</code></td>
<td>Generates DS record(s) to be included in the parent zone</td>
</tr>
<tr>
<td><code>-s start-time</code></td>
<td>Sets the time at which the signatures become valid</td>
</tr>
<tr>
<td><code>-e end-time</code></td>
<td>Sets the time at which the signatures expire</td>
</tr>
<tr>
<td><code>-t</code></td>
<td>Prints statistics</td>
</tr>
</tbody>
</table>
The dates and times for signature validity can be expressed as absolute times in the format `yyyymmddhhmmss` or as times relative to now in the format `+N`, where `N` is in seconds. The default signature validity period is from an hour in the past to 30 days in the future. Here is an example in which we specify that signatures should be valid until the end of the calendar year 2017:

```sh
$ sudo dnssec-signzone -N increment -e 20171231235959 example.com
```

Signed zone files are typically four to ten times larger than the original zone, and all your nice logical ordering is lost. A line such as

```text
mail-relay A 63.173.189.2
```

becomes several lines:

```text
mail-relay.example.com. 57600 A 63.173.189.2
  57600 RRSIG A 8 3 57600 20090722234636 (  
    20150622234636 23301 example.com. 
    Y7s9jDWyuUvozeU7zGRdFCl+rzU8cLiwoev 
    012TGlblhsRgJfkpEYFVRUB7kKVRNgueYwk 
    d2RSD9q9zRQ+w== )

3600 NSEC mail-relay2.example.com. A RRSIG NSEC  
3600 RRSIG NSEC 8 3 3600 20090722234636 (  
    20150622234636 23301 example.com. 
    42QrXP8vpoChsGPseProBMZ7twf7e55WK+40 
    WNsN84hF0notymRzRIZypqWzlIPBZAUJ77R 
    HP0hlfBDqwmZYw== )
```

In practical terms, a signed zone file is no longer human-readable, and it cannot be edited by hand because of the RRSIG and NSEC or NSEC3 records. No user-serviceable parts inside!

With the exception of DNSKEY records, each resource record set (resource records of the same type for the same name) gets one signature from the ZSK. DNSKEY resource records are signed by both the ZSK and the KSK, so they have two RRSIGs. The base-64 representation of a signature ends in however many equal signs are needed to make the length a multiple of 4.

Once your zones are signed, all that remains is to point your name server at the signed versions of the zone files. If you’re using BIND, look for the `zone` statement that corresponds to each zone in `named.conf` and change the `file` parameter from `example.com` to `example.com.signed`.

Finally, restart the name server daemon, telling it to reread its configuration file with `sudo rndc reconfig` followed by `sudo rndc flush`.

You are now serving a DNSSEC signed zone! To make changes, you can edit either the original unsigned zone or the signed zone and then re-sign the zone. Editing a signed zone is something of a logistical nightmare, but it is much quicker than re-signing the entire zone. Be sure to
remove the RRSIG records that correspond to any records that you change. You probably want to make identical changes to the unsigned zone to avoid version skew.

If you pass a signed zone as the argument to `dnssec-signzone`, any unsigned records are signed and the signatures of any records that are close to expiring are renewed. “Close to expiring” is defined as being three-quarters of the way through the validity period. Re-signing typically results in changes, so make sure you increment the zone’s serial number by hand or use `dnssec-signzone -N increment` to automatically increment the zone’s serial number.

That’s all there is to the local part of DNSSEC configuration. What’s left is the thorny problem of getting your island of secure DNS connected to other trusted, signed parts of the DNS archipelago.
The DNSSEC chain of trust

Continuing with our example DNSSEC setup, example.com is now signed and its name servers have DNSSEC enabled. This means that when querying they use EDNS0, the extended DNS protocol, and set the DNSSEC-aware option in the DNS header of the packet. When answering a query that arrives with that bit set, they include the signature data with their answer.

A client that receives signed answers can validate the response by checking the record’s signatures with the appropriate public key. But it gets this key from the zone’s own DNSKEY record, which is rather suspicious if you think about it. What’s to stop an impostor from serving up both fake records and a fake key that validates them?

The canonical solution is that you give your parent zone a DS record to include in its zone file. By virtue of coming from the parent zone, the DS record is certified by the parent’s private key. If the client trusts your parent zone, it should then trust that the parent zone’s DS record accurately reflects your zone’s public key.

The parent zone is in turn certified by its parent, and so on back to the root.
**DNSSEC key rollover**

Key rollover has always been a troublesome issue in DNSSEC. In fact, the original specifications were changed specifically to address the issue of the communication needed between parent and child zones whenever keys were created, changed, or deleted. The new specifications are called DNSSEC-bis.

ZSK rollover is relatively straightforward and does not involve your parent zone or any trust anchor issues. The only tricky part is the timing. Keys have an expiration time, so rollover must occur well before that time. However, keys also have a TTL, defined in the zone file. To illustrate, assume that the TTL is one day and that keys don’t expire for another week. The following steps are then involved:

1. Generate a new ZSK.
2. Include it in the zone file.
3. Sign or re-sign the zone with the KSK and the *old* ZSK.
4. Signal the name server to reload the zone; the new key is now there.
5. Wait 24 hours (the TTL); now everyone has both the old and new keys.
6. Sign the zone again with the KSK and the *new* ZSK.
7. Signal the name server to reload the zone.
8. Wait another 24 hours; now everyone has the new signed zone.
9. Remove the old ZSK at your leisure, e.g., the next time the zone changes.

This scheme is called prepublishing. Obviously, you must start the process at least two TTLs before the point at which you need to have everyone using the new key. The waiting periods guarantee that any site with cached values always has a cached key that corresponds to the cached data.

Another variable that affects this process is the time it takes for your slowest slave server to update its copy of your zone when notified by the master server. So don’t wait until the last minute to start your rollover process or to re-sign zones whose signatures are expiring. Expired signatures do not validate, so sites that verify DNSSEC signatures will not be able to do DNS lookups for your domain.

The mechanism to roll over a KSK is called double signing and it’s also pretty straightforward. However, you will need to communicate your new DS record to your parent. Make sure you have positive acknowledgement from the parent before you switch to just the new key. Here are the steps:

1. Create a new KSK.
2. Include it in the zone file.
3. Sign the zone with both old and new KSKs and the ZSK.
4. Signal the name server to reload the zone.
5. Wait 24 hours (the TTL); now everyone has the new key.
6. After confirmation, delete the old KSK record from the zone.
7. Re-sign the zone with the new KSK and ZSK.
DNSSEC tools

With the advent of BIND 9.10 comes a new debugging tool. The Domain Entity Lookup and Validation engine (DELV) looks much like dig but has a better understanding of DNSSEC. In fact, delv checks the DNSSEC validation chain with the same code that is used by the BIND 9 named itself.

In addition to the DNSSEC tools that come with BIND, four other deployment and testing toolsets might be helpful: ldns, DNSSEC-Tools (formerly Sparta), RIPE, and OpenDNSSEC (opendnssec.org).

ldns tools, nlnetlabs.nl/projects/ldns

ldns, from the folks at NLnet Labs, is a library of routines for writing DNS tools and a set of example programs that use this library. We list the tools and what each one does below. The tools are all in the examples directory except for drill, which has its own directory in the distribution. Man pages can be found with the commands. The top-level README file gives very brief installation instructions.

- ldns-chaos shows the name server ID info stored in the CHAOS class.
- ldns-compare-zones shows the differences between two zone files.
- ldns-dpa analyzes DNS packets in tcpdump trace files.
- ldns-key2ds converts a DNSKEY record to a DS record.
- ldns-keyfetcher fetches DNSSEC public keys for zones.
- ldns-keygen generates TSIG keys and DNSSEC key pairs.
- ldns-notify makes a zone’s slave servers check for updates.
- ldns-nsec3-hash prints the NSEC3 hash for a name.
- ldns-read-zone reads a zone and prints it in various formats.
- ldns-revoke sets the revoke flag on a DNSKEY key RR (RFC5011).
- ldns-rrsig prints human-readable expiration dates from RRSIGs.
- ldns-signzone signs a zone file with either NSEC or NSEC3.
- ldns-update sends a dynamic update packet.
- ldns-verify-zone makes sure RRSIG, NSEC, and NSEC3 records are OK.
- ldns-walk walks through a zone by following the DNSSEC NSEC records.
- **ldns-zcat** reassembles zone files split with **ldns-zsplit**.
- **ldns-zsplit** splits a zone into chunks so it can be signed in parallel.

Many of these tools are simple and do only one tiny DNS chore. They were written as example uses of the **ldns** library and demonstrate how simple the code becomes when the library does all the hard bits for you.

**dnssec-tools.org**

DNSSec-tools builds on the BIND tools and includes the following commands:

- **dnspktflow** traces the flow of DNS packets during a query/response sequence captured by **tcpdump** and produces a cool diagram.
- **donuts** analyzes zone files and finds errors and inconsistencies.
- **donutsd** runs **donuts** at intervals and warns of problems.
- **mapper** maps zone files, showing secure and insecure portions.
- **rollerd**, **rollctl**, and **rollinit** automate key rollovers by using the prepublishing scheme for ZSKs and the double signature method for KSKs. See [this page](#) for the details of these schemes.
- **trustman** manages trust anchors and includes an implementation of RFC5011 key rollover.
- **validate** validates signatures from the command-line.
- **zonesigner** generates keys and signs zones.

The web site contains good documentation and tutorials for all of these tools. The source code is available for download and is covered by the BSD license.

**RIPE tools, ripe.net**

RIPE’s tools act as a front end to BIND’s DNSSEC tools and focus on key management. They have friendlier messages since they run and package up the many arguments and commands into more intuitive forms.

**OpenDNSSEC, opendnssec.org**

OpenDNSSEC is a set of tools that takes unsigned zones, adds the signatures and other records for DNSSEC, and passes it on to the authoritative name servers for that zone. This automation greatly simplifies the initial setup of DNSSEC.
Debugging DNSSEC

DNSSEC interoperates with both signed and unsigned zones, and with both DNSSEC-aware and DNSSEC-oblivious name servers. Ergo, incremental deployment is possible, and it usually just works. But not always.

DNSSEC is a distributed system with lots of moving parts. Servers, resolvers, and the paths among them can all experience problems. A problem seen locally may originate far away, so tools like SecSpider and Vantages that monitor the distributed state of the system can be helpful. Those tools, the utilities mentioned in the previous section, and your name server log files are your primary debugging weapons.

Make sure that you route the DNSSEC logging category in named.conf to a file on the local machine. It’s helpful to separate out the DNSSEC-related messages so that you don’t route any other logging categories to this file. Here is an example logging specification for named:

```plaintext
channel dnssec-log {    
    file "/var/log/named/dnssec.log" versions 4 size 10m;    
    print-time yes;    
    print-category yes;    
    print-severity yes;    
    severity debug 3;    
};

category dnssec { dnssec-log; };
```

In BIND, set the debugging level to 3 or higher to see the validation steps taken by a recursive BIND server trying to validate a signature. This logging level produces about two pages of logging output per signature verified. If you are monitoring a busy server, log data from multiple queries will likely be interleaved. Sorting through the mess can be challenging and tedious.

drill has two particularly useful flags: -T to trace the chain of trust from the root to a specified host, and -S to chase the signatures from a specified host back to the root. Here’s some made-up sample output from drill -S snitched from the DNSSEC HOWTO at NLnet Labs:

```bash
$ drill -S -k ksk.keyfile example.net SOA
DNSSEC Trust tree:
example.net. (SOA)
|---.example.net. (DNSKEY keytag: 17000)
  |---.example.net. (DNSKEY keytag: 49656)
  |---.example.net. (DS keytag: 49656)
  |---.net. (DNSKEY keytag: 62972)
    |---.net. (DNSKEY keytag: 13467)
    |---.net. (DS keytag: 13467)
    |---. (DNSKEY keytag: 63380)
      |---. (DNSKEY keytag: 63276) ;; Chase successful
```
If a validating name server cannot verify a signature, it returns a SERVFAIL indication. The underlying problem could be a configuration error by someone at one of the zones in the chain of trust, bogus data from an interloper, or a problem in the setup of the validating recursive server itself. Try **drill** to chase the signatures along the chain of trust and see where the problem lies.

If all the signatures are verified, try querying the troublesome site with **dig** and then with **dig +cd**. (The **cd** flag turns off validation.) Try this at each of the zones in the chain of trust to see if you can find the problem. You can work your way up or down the chain of trust. The likely result will be an expired trust anchor or expired signatures.
16.11 BIND DEBUGGING

BIND provides three basic debugging tools: logging, described below; a control program, described starting here; and a command-line query tool, described here.
Logging in BIND

See Chapter 10 for more information about syslog.

named’s logging facilities are flexible enough to make your hair stand on end. BIND originally just used syslog to report error messages and anomalies. Recent versions generalize the syslog concepts by adding another layer of indirection and support for logging directly to files. Before you dive in, check the mini-glossary of BIND logging terms shown in Table 16.7.

Table 16.7: A BIND logging lexicon

<table>
<thead>
<tr>
<th>Term</th>
<th>What it means</th>
</tr>
</thead>
<tbody>
<tr>
<td>category</td>
<td>A class of messages that named can generate; for example, messages about dynamic updates or messages about answering queries</td>
</tr>
<tr>
<td>module</td>
<td>The name of the source module that generates a message</td>
</tr>
<tr>
<td>severity</td>
<td>The “badness” of an error message; what syslog refers to as a priority</td>
</tr>
<tr>
<td>channel</td>
<td>A place where messages can go: syslog, a file, or /dev/null^</td>
</tr>
<tr>
<td>facility</td>
<td>A syslog facility name. DNS does not have its own specific facility, but you have your pick of all the standard ones</td>
</tr>
</tbody>
</table>

^ /dev/null is a pseudo-device that throws away all input.

You configure BIND logging with a logging statement in named.conf. You first define channels, the possible destinations for messages. You then direct various categories of message to go to particular channels.

When a message is generated, it is assigned a category, a module, and a severity at its point of origin. It is then distributed to all the channels associated with its category and module. Each channel has a severity filter that tells what severity level a message must have to get through. Channels that lead to syslog stamp messages with the designated facility name. Messages that go to syslog are also filtered according to the rules in /etc/syslog.conf. Here’s the outline of a logging statement:

```plaintext
logging {
    channel-def;
    channel-def;
    ...
    category category-name {
        channel-name;
        channel-name;
        ...
    };
};
```
Channels

A channel-def looks slightly different according to whether the channel is a file channel or a syslog channel. You must choose file or syslog for each channel; a channel can’t be both at the same time.

channel channel-name {
    file path [versions numvers | unlimited] [size sizespec];
    syslog facility;
    severity severity;
    print-category yes | no;
    print-severity yes | no;
    print-time yes | no;
};

For a file channel, numvers tells how many backup versions of a file to keep, and sizespec specifies how large the file should be allowed to grow (examples: 2048, 100k, 20m, unlimited, default) before it is automatically rotated. If you name a file channel mylog, the rotated versions are mylog.0, mylog.1, and so on.

See this page for a list of syslog facility names.

In the syslog case, facility names the syslog facility under which to log the message. It can be any standard facility. In practice, only daemon and local0 through local7 are reasonable choices.

The rest of the statements in a channel-def are optional. severity can have the values (in descending order) critical, error, warning, notice, info, or debug (with an optional numeric level, e.g., severity debug 3). The value dynamic is also recognized and matches the server’s current debug level.

The various print options add or suppress message prefixes. Syslog prepends the time and reporting host to each message logged, but not the severity or the category. The source filename (module) that generated the message is also available as a print option. It makes sense to enable print-time only for file channels—syslog adds its own time stamps, so there’s no need to duplicate them.

The four channels listed in Table 16.8 are predefined by default. These defaults should be fine for most installations.

Table 16.8: Predefined logging channels in BIND
Categories

Categories are determined by the programmer at the time the code is written. They organize log messages by topic or functionality instead of just by severity. Table 16.9 shows the current list of message categories.

Table 16.9: BIND logging categories

<table>
<thead>
<tr>
<th>Category</th>
<th>What it includes</th>
</tr>
</thead>
<tbody>
<tr>
<td>client</td>
<td>Client requests</td>
</tr>
<tr>
<td>config</td>
<td>Configuration file parsing and processing</td>
</tr>
<tr>
<td>database</td>
<td>Messages about database operations</td>
</tr>
<tr>
<td>default</td>
<td>Default for categories without specific logging options</td>
</tr>
<tr>
<td>delegation-only</td>
<td>Queries forced to NXDOMAIN by delegation-only zones</td>
</tr>
<tr>
<td>dispatch</td>
<td>Dispatching of incoming packets to server modules</td>
</tr>
<tr>
<td>dnssec</td>
<td>DNSSEC messages</td>
</tr>
<tr>
<td>edns-disabled</td>
<td>Info about broken servers</td>
</tr>
<tr>
<td>general</td>
<td>Catchall for unclassified messages</td>
</tr>
<tr>
<td>lame-servers</td>
<td>Servers that are supposed to be serving a zone, but aren’t(^a)</td>
</tr>
<tr>
<td>network</td>
<td>Network operations</td>
</tr>
<tr>
<td>notify</td>
<td>Messages about the “zone changed” notification protocol</td>
</tr>
<tr>
<td>queries</td>
<td>A short log message for every query the server receives (!)</td>
</tr>
<tr>
<td>resolver</td>
<td>DNS resolution, e.g., recursive lookups for clients</td>
</tr>
<tr>
<td>security</td>
<td>Approved/unapproved requests</td>
</tr>
<tr>
<td>unmatched</td>
<td>Queries named cannot classify (bad class, no view)</td>
</tr>
<tr>
<td>update</td>
<td>Messages about dynamic updates</td>
</tr>
<tr>
<td>update-security</td>
<td>Approval or denial of update requests</td>
</tr>
<tr>
<td>xfer-in</td>
<td>Zone transfers that the server is receiving</td>
</tr>
<tr>
<td>xfer-out</td>
<td>Zone transfers that the server is sending</td>
</tr>
</tbody>
</table>

\(^a\) Either the parent zone or the child zone could be at fault; hard to tell without investigating.

Log messages

The default logging configuration is
You should watch the log files when you make major changes to BIND and perhaps increase the logging level. Later, reconfigure to preserve only serious messages once you have verified that named is stable.

Query logging can be quite educational. You can verify that your allow clauses are working, see who is querying you, identify broken clients, etc. It’s a good check to perform after major reconfigurations, especially if you have a good sense of what your query load looked like before the changes.

To start query logging, just direct the queries category to a channel. Writing to syslog is less efficient than writing directly to a file, so use a file channel on a local disk when you are logging every query. Have lots of disk space and be ready to turn query logging off once you obtain enough data. (rndc querylog dynamically toggles query logging on and off.)

Views can be pesky to debug, but fortunately, the view that matched a particular query is logged along with the query.

Some common log messages are listed below:

- **Lame server resolving xxx.** If you get this message about one of your own zones, you have configured something incorrectly. The message is harmless if it’s about some zone out on the Internet; it’s someone else’s problem. A good one to throw away by directing it to the null channel.

- **…query (cache) xxx denied.** This can be either misconfiguration of the remote site, abuse, or a case in which someone has delegated a zone to you, but you have not configured it.

- **Too many timeouts resolving xxx: disabling EDNS.** This message can result from a broken firewall not admitting UDP packets over 512 bytes long or not admitting fragments. It can also be a sign of problems at the specified host. Verify that the problem is not your firewall and consider redirecting these messages to the null channel.

- **Unexpected RCODE (SERVFAIL) resolving xxx.** This can be an attack or, more likely, a sign of something repeatedly querying a lame zone.

- **Bad referral.** This message indicates a miscommunication among a zone’s name servers.

- **Not authoritative for.** A slave server is unable to get authoritative data for a zone. Perhaps it’s pointing to the wrong master, or perhaps the master had trouble loading
the zone in question.

- **Rejected zone.** `named` rejected a zone file because it contained errors.

- **No NS RRs found.** A zone file did not include NS records after the SOA record. It could be that the records are missing, or it could be that they don’t start with a tab or other whitespace. In the latter case, the records are not attached to the zone of the SOA record and are therefore misinterpreted.

- **No default TTL set.** The preferred way to set the default TTL for resource records is with a `$TTL` directive at the top of the zone file. This error message indicates that the `$TTL` is missing; it is required in BIND 9.

- **No root name server for class.** Your server is having trouble finding the root name servers. Check your hints file and the server’s Internet connectivity.

- **Address already in use.** The port on which `named` wants to run is already being used by another process, probably another copy of `named`. If you don’t see another `named` around, it might have crashed and left an `rndc` control socket open that you’ll have to track down and remove. A good way to fix the problem is to stop the `named` process with `rndc` and then restart `named`:

  ```
  $ sudo rndc stop
  $ sudo /usr/sbin/named ...
  ```

- **...updating zone xxx: update unsuccessful.** A dynamic update for a zone was attempted but refused, most likely because of the `allow-update` or `update-policy` clause in `named.conf` for this zone. This is a common error message and often is caused by misconfigured Windows boxes.

### Sample BIND logging configuration

The following snippet from the ISC `named.conf` file for a busy TLD name server illustrates a comprehensive logging regimen.
Debug levels in BIND

**named** debug levels are denoted by integers from 0 to 100. The higher the number, the more verbose the output. Level 0 turns debugging off. Levels 1 and 2 are fine for debugging your configuration and database. Levels beyond about 4 are appropriate for the maintainers of the code.

You invoke debugging on the **named** command line with the `-d` flag. For example,

```
$ sudo named -d2
```

would start **named** at debug level 2. By default, debugging information is written to the file `named.run` in the current working directory from which **named** is started. The `named.run` file grows fast, so don’t go out for a beer while debugging or you might have bigger problems when you return.

You can also turn on debugging while **named** is running with `rndc trace`, which increments the debug level by 1, or with `rndc trace level`, which sets the debug level to the value specified. `rndc notrace` turns debugging off completely. You can also enable debugging by defining a logging channel that includes a severity specification such as

```plaintext
logging {
  channel default-log { # Default channel, to a file
    file "log/named.log" versions 3 size 10m;
    print-time yes;
    print-category yes;
    print-severity yes;
    severity info;
  };
  channel xfer-log { # Zone transfers channel, to a file
    file "log/xfer.log" versions 3 size 10m;
    print-category yes;
    print-severity yes;
    print-time yes;
    severity info;
  };
  channel dnssec-log { # DNSSEC channel, to a file
    file "log/dnssec.log" versions 3 size 1M;
    severity debug 1;
    print-severity yes;
    print-time yes;
  };
  category default { default-log; default_debug; };
  category dnssec { dnssec-log; };
  category xfer-in { xfer-log; };
  category xfer-out { xfer-log; };
  category notify { xfer-log; };
};
```
severity debug 3;

which sends all debugging messages up to level 3 to that particular channel. Other lines in the channel definition specify the destination of those debugging messages. The higher the severity level, the more information is logged.

Watching the logs or the debugging output illustrates how often DNS is misconfigured in the real world. That pesky little dot at the end of names (or rather, the lack thereof) accounts for an alarming amount of DNS traffic.
Name server control with rndc

Table 16.10 shows some of the options accepted by **rndc**. Typing **rndc** with no arguments lists the available commands and briefly describes what they do. Earlier incantations of **rndc** used signals, but with over 25 commands, the BIND folks ran out of signals long ago. Commands that produce files put them in whatever directory is specified as **named**’s home in **named.conf**.

### Table 16.10: rndc commands

<table>
<thead>
<tr>
<th>Command</th>
<th>Function</th>
</tr>
</thead>
<tbody>
<tr>
<td>dumpdb</td>
<td>Dumps the DNS database to named_dump.db</td>
</tr>
<tr>
<td>flush [view]</td>
<td>Flushes all caches or those for a specified view</td>
</tr>
<tr>
<td>flushname name [view]</td>
<td>Flushes the specified name from the server’s cache</td>
</tr>
<tr>
<td>freeze zone [class [view]]</td>
<td>Suspends updates to a dynamic zone</td>
</tr>
<tr>
<td>thaw zone [class [view]]</td>
<td>Resumes updates to a dynamic zone</td>
</tr>
<tr>
<td>halt</td>
<td>Halts named without writing pending updates</td>
</tr>
<tr>
<td>querylog</td>
<td>Toggles tracing of incoming queries</td>
</tr>
<tr>
<td>notify zone [class [view]]</td>
<td>Resends notification messages for zone</td>
</tr>
<tr>
<td>notrace</td>
<td>Turns off debugging</td>
</tr>
<tr>
<td>reconfig</td>
<td>Reloads the config file and loads any new zones</td>
</tr>
<tr>
<td>recursing</td>
<td>Dumps queries currently recursing, named.recursing</td>
</tr>
<tr>
<td>refresh zone [class [view]]</td>
<td>Schedules maintenance for a zone</td>
</tr>
<tr>
<td>reload</td>
<td>Reloads named.conf and zone files</td>
</tr>
<tr>
<td>reload zone [class [view]]</td>
<td>Reloads only the specified zone or view</td>
</tr>
<tr>
<td>retransfer zone [class [view]]</td>
<td>Recopies the data for zone from the master server</td>
</tr>
<tr>
<td>stats</td>
<td>Dumps statistics to named.stats</td>
</tr>
<tr>
<td>status</td>
<td>Displays the current status of the running named</td>
</tr>
<tr>
<td>stop</td>
<td>Saves pending updates and then stops named</td>
</tr>
<tr>
<td>trace</td>
<td>Increments the debug level by 1</td>
</tr>
<tr>
<td>trace level</td>
<td>Changes the debug level to the value level</td>
</tr>
<tr>
<td>validation newstate</td>
<td>Enables/disables DNSSEC validation on the fly</td>
</tr>
</tbody>
</table>

a. The class argument here is the same as for resource records, typically IN for Internet.

**rndc reload** makes **named** reread its configuration file and reload zone files. The **reload** zone command is handy when only one zone has changed and you don’t want to reload all the zones, especially on a busy server. You can also specify a class and view to reload only the selected view of the zone’s data.

Note that **rndc reload** is not sufficient to add a completely new zone; that requires **named** to read both the **named.conf** file and the new zone file. For new zones, use **rndc reconfig**, which rereads the config file and loads any new zones without disturbing existing zones.
**rndc freeze zone** stops dynamic updates and reconciles the journal of dynamic updates to the data files. After freezing the zone, you can edit the zone data by hand. As long as the zone is frozen, dynamic updates are refused. Once you’ve finished editing, use **rndc thaw zone** to start accepting dynamic updates again.

**rndc dumpdb** instructs **named** to dump its database to **named_dump.db**. The dump file is big and includes not only local data but also any cached data the name server has accumulated.

Your versions of **named** and **rndc** must match or you will see an error message about a protocol version mismatch. They’re normally installed together on individual machines, but version skew can be an issue when you are trying to control a **named** on another computer.
Command-line querying for lame delegations

When you apply for a domain name, you are asking for a part of the DNS naming tree to be delegated to your name servers and your DNS administrator. If you never use the domain or you change the name servers or their IP addresses without coordinating with your parent zone, a “lame delegation” results.

The effects of a lame delegation can be really bad. If one of your servers is lame, your DNS system is less efficient. If all the name servers for a domain are lame, no one can reach you. All queries start at the root unless answers are cached, so lame servers and lazy software that doesn’t do negative caching of SERVFAIL errors increase the load of everyone on the path from the root to the lame domain.

The doc (“domain obscenity control”) command can help you identify lame delegations, but you can also find them just by reviewing your log files. Here’s an example log message:

```
Jul 19 14:37:50 nubark named[757]: lame server resolving 'w3w3.com'
   (in 'w3w3.com'?): 216.117.131.52#53
```

Digging for name servers for w3w3.com at one of the .com gTLD servers yields the results below. We have truncated the output to tame `dig`’s verbosity; the `+short` flag to `dig` limits the output even more.

```
$ dig @e.gtld-servers.net w3w3.com ns
;; ANSWER SECTION:
w3w3.com. 172800 IN NS ns0.nameservices.net.
w3w3.com. 172800 IN NS ns1.nameservices.net.
```

If we query each of these servers in turn, we get an answer from ns0 but not from ns1:

```
$ dig @ns0.nameservices.net w3w3.com ns
;; ANSWER SECTION:
w3w3.com. 14400 IN NS ns0.nameservices.net.
w3w3.com. 14400 IN NS ns1.nameservices.net.
```

```
$ dig @ns1.nameservices.net w3w3.com ns
;; QUESTION SECTION:
;w3w3.com. IN NS

;; AUTHORITY SECTION:
com. 92152 IN NS M.GTLD-SERVERS.NET.
com. 92152 IN NS I.GTLD-SERVERS.NET.
com. 92152 IN NS E.GTLD-SERVERS.NET.
```

The server ns1.nameservices.net has been delegated responsibility for w3w3.com by the .com servers, but it does not accept that responsibility. It is misconfigured, resulting in a lame delegation. Clients trying to look up w3w3.com will experience slow service. If w3w3.com is paying nameservices.net for DNS service, they deserve a refund!
Many sites point their lame-servers logging channel to /dev/null and don’t fret about other people’s lame delegations. That’s fine as long as your own domain is squeaky clean and is not itself a source or victim of lame delegations.

Sometimes when you dig at an authoritative server in an attempt to find lameness, dig returns no information. Try the query again with the +norecurse flag so that you can see exactly what the server in question knows.
16.12 RECOMMENDED READING

DNS and BIND are described by a variety of sources, including the documentation that comes with the distributions, chapters in several books on Internet topics, books in the O’Reilly Nutshell series, books from other publishers, and various on-line resources.
**Books and other documentation**

**THE NOMINUM AND ISC BIND DEVELOPMENT TEAMS.** *BIND 9 Administrator Reference Manual.* This manual is included in the BIND distribution ([doc/arm](doc/arm)) from isc.org and is also available separately from the same site. It outlines the administration and management of BIND 9.

**LIU, CRICKET, AND PAUL ALBITZ.** *DNS and BIND (5th Edition).* Sebastopol, CA: O’Reilly Media, 2006. This is pretty much the BIND bible, although it’s getting a bit long in the tooth.

**LIU, CRICKET.** *DNS & BIND Cookbook.* Sebastopol, CA: O’Reilly Media, 2002. This baby version of the O’Reilly DNS book is task oriented and gives clear instructions and examples for various name server chores. Dated, but still useful.

**LIU, CRICKET.** *DNS and BIND on IPv6.* Sebastopol, CA: O’Reilly Media, 2011. This is an IPv6-focused addendum to DNS and BIND. It’s short and includes only IPv6-related material.

On-line resources

The web sites isc.org, dns-oarc.net, ripe.net, and nlnetlabs.nl contain a wealth of DNS information, research, measurement results, presentations, and other good stuff.

All the nitty-gritty details of the DNS protocol, resource records, and the like are summarized at iana.org/assignments/dns-parameters. This document contains a nice mapping from a DNS fact to the RFC that specifies it.

The DNSSEC HOWTO, a tutorial in disguise by Olaf Kolkman, is a 70-page document that covers the ins and outs of deploying and debugging DNSSEC. Get it at nlnetlabs.nl/dnssec_howto/dnssec_howto.pdf.
The RFCs

The RFCs that define the DNS system are available from rfc-editor.org. We formerly listed a page or so of the most important DNS-related RFCs, but there are now so many (more than 100, with another 50 Internet drafts) that you are better off searching rfc-editor.org to access the entire archive.

Refer to the doc/rfc and doc/draft directories of the current BIND distribution to see the entire complement of DNS-related RFCs.
Both users and system administrators would like account information to magically propagate to all an environment’s computers so that a user can log in to any system with the same credentials. The common term for this feature is “single sign-on” (SSO), and the need for it is universal.

SSO involves two core security concepts: identity and authentication. A user identity is the abstract representation of an individual who needs access to a system or an application. It typically includes attributes such as a username, password, user ID, and email address. Authentication is the act of proving that an individual is the legitimate owner of an identity.

This chapter focuses on SSO as a component of UNIX and Linux systems within a single organization. For interorganizational SSO (such as might be needed to integrate your systems with a Software-as-a-Service provider), several standards-based and commercial SSO solutions are available. For those cases, we recommend learning about Security Assertion Markup Language (SAML) as a first step on your journey.
17.1 Core SSO Elements

Although there are many ways to set up SSO, four elements are typically required in every scenario:

- A centralized directory store that contains user identity and authorization information. The most common solutions are directory services based on the Lightweight Directory Access Protocol (LDAP). In environments that mix Windows, UNIX, and Linux systems, the ever-popular Microsoft Active Directory service is a good choice. Active Directory includes a customized, nonstandard LDAP interface.

- A tool for managing user information in the directory. For native LDAP implementations, we recommend phpLDAPadmin or Apache Directory Studio. Both are easy-to-use, web-based tools that let you import, add, modify, and delete directory entries. If you’re a Microsoft Active Directory fan-person, you can use the Windows-native MMC snap-in “Active Directory Users and Computers” to manage information in the directory.

- A mechanism for authenticating user identities. You can authenticate users directly against an LDAP store, but it’s also common to use the Kerberos ticket-based authentication system originally developed at MIT. In Windows environments, Active Directory supplies LDAP access to user identities and uses a customized version of Kerberos for authentication.

Authentication on modern UNIX and Linux systems goes through the Pluggable Authentication Module system, aka PAM. You can use the System Security Services Daemon (sssd) to aggregate access to user identity and authentication services, then point PAM at sssd.

- Centralized-identity-and-authentication-aware versions of the C library routines that look up user attributes. These routines (e.g., getpwent) historically read flat files such as /etc/passwd and /etc/group and answered queries from their contents. These days, the data sources are configured in the name service switch file, /etc/nsswitch.conf.

The security community is divided over whether authentication is most secure when performed through LDAP or Kerberos. The road of life is paved with flat squirrels that couldn’t decide. Pick an option and don’t look back.

Exhibit A illustrates the high-level relationships of the various components in a typical configuration. This example uses Active Directory as the directory server. Note that both time synchronization (NTP) and hostname mapping (DNS) are critical for environments that use Kerberos because authentication tickets are time stamped and have a limited validity period.
In this chapter, we cover core LDAP concepts and introduce two specific LDAP servers for UNIX and Linux. We then discuss the steps needed to make a machine use a centralized directory service to process logins.
17.2 LDAP: “LIGHTWEIGHT” DIRECTORY SERVICES

A directory service is just a database, but one that makes a few assumptions. Any kind of data that matches the assumptions is a candidate for inclusion in the directory. The basic assumptions are as follows:

- Data objects are relatively small.
- The database will be widely replicated and cached.
- The information is attribute-based.
- Data are read often but written infrequently.
- Searching is a common operation.

The current IETF standards-track protocol that fills this role is the Lightweight Directory Access Protocol (LDAP). Ironically, LDAP is anything but lightweight. It was originally a gateway protocol that allowed TCP/IP clients to talk to an older directory service called X.500, which is now obsolete.

Microsoft’s Active Directory is the most common instantiation of LDAP, and many sites use Active Directory for both Windows and UNIX/Linux authentication. For environments in which Active Directory isn’t a candidate, the OpenLDAP package (openldap.org) has become the standard implementation. The 389 Directory Server (formerly known as the Fedora Directory Server and the Netscape Directory Server) is also open source and can be found at port389.org. The name derives from the fact that TCP port 389 is the default port for all LDAP implementations.
Uses for LDAP

Until you’ve had some experience with it, LDAP can be a slippery fish to grab hold of. LDAP by itself doesn’t solve any specific administrative problem. Today, the most common use of LDAP is to act as a central repository for login names, passwords, and other account attributes. However, LDAP can be used in many other ways:

- LDAP can store additional directory information about users, such as phone numbers, home addresses, and office locations.
- Most mail systems—including sendmail, Exim, and Postfix—can draw a large part of their routing information from LDAP. See this page for more information about using LDAP with sendmail.
- LDAP makes it easy for applications (even those written by other teams and departments) to authenticate users without having to worry about the exact details of account management.
- LDAP is well supported by common scripting languages such as Perl and Python through code libraries. Ergo, LDAP can be an elegant way to distribute configuration information for locally written scripts and administrative utilities.
- LDAP is well supported as a public directory service. Most major email clients can read user directories stored in LDAP. Simple LDAP searches are also supported by many web browsers through an LDAP URL type.
The structure of LDAP data

LDAP data takes the form of property lists, which are known in the LDAP world as “entries.” Each entry consists of a set of named attributes (such as description or uid) along with those attributes’ values. Every attribute can have multiple values. Windows users might recognize this structure as being similar to that of the Windows registry.

As an example, here’s a typical (but simplified) /etc/passwd line expressed as an LDAP entry:

dn: uid=ghopper,ou=People,dc=navy,dc=mil
objectClass: top
objectClass: person
objectClass: organizationalPerson
objectClass: inetOrgPerson
objectClass: posixAccount
objectClass: shadowAccount
uid: ghopper
cn: Grace Hopper
userPassword: {crypt}$1$pZaGA2RL$MPDJoc0afuhHY6yk8HQPp0
loginShell: /bin/bash
uidNumber: 1202
gidNumber: 1202
homeDirectory: /home/ghopper

This notation is a simple example of LDIF, the LDAP Data Interchange Format, which is used by most LDAP-related tools and server implementations. The fact that LDAP data can be easily converted back and forth from plain text is part of the reason for its success.

Entries are organized into a hierarchy through the use of “distinguished names” (attribute name: dn) that form a sort of search path. As in DNS, the “most significant bit” goes on the right. In the example above, the DNS name navy.mil has structured the top levels of the LDAP hierarchy. It has been broken down into two domain components (dc’s), “navy” and “mil,” but this is only one of several common conventions.

Every entry has exactly one distinguished name. Entries are entirely separate from one another and have no hierarchical relationship except as is implicitly defined by the dn attributes. This approach enforces uniqueness and gives the implementation a hint as to how to efficiently index and search the data. Various LDAP consumers use the virtual hierarchy defined by dn attributes, but that’s more a data-structuring convention than an explicit feature of the LDAP system. There are, however, provisions for symbolic links between entries and for referrals to other servers.

LDAP entries are typically schematized through the use of an objectClass attribute. Object classes specify the attributes that an entry can contain, some of which may be required for validity. The schemata also assign a data type to each attribute. Object classes nest and combine in the traditional object-oriented fashion. The top level of the object class tree is the class named top, which specifies merely that an entry must have an objectClass attribute.
Table 17.1 shows some common LDAP attributes whose meanings might not be immediately apparent. These attributes are case-insensitive.

Table 17.1: Some common attribute names found in LDAP hierarchies

<table>
<thead>
<tr>
<th>Attribute</th>
<th>Stands for</th>
<th>What it is</th>
</tr>
</thead>
<tbody>
<tr>
<td>o</td>
<td>Organization</td>
<td>Often identifies a site's top-level entry *</td>
</tr>
<tr>
<td>ou</td>
<td>Organizational unit</td>
<td>A logical subdivision, e.g., “marketing”</td>
</tr>
<tr>
<td>cn</td>
<td>Common name</td>
<td>The most natural name to represent the entry</td>
</tr>
<tr>
<td>dc</td>
<td>Domain component</td>
<td>Used at sites that model their hierarchy on DNS</td>
</tr>
<tr>
<td>objectClass</td>
<td>Object class</td>
<td>Schema to which this entry's attributes conform</td>
</tr>
</tbody>
</table>

*a. Typically not used by sites that model their LDAP hierarchy on DNS*
OpenLDAP: the traditional open source LDAP server

OpenLDAP is an extension of work originally done at the University of Michigan; it now continues as an open source project. It’s shipped with most Linux distributions, although it is not necessarily included in the default installation. The documentation is perhaps best described as “brisk.”

In the OpenLDAP distribution, slapd is the standard LDAP server daemon. In an environment with multiple OpenLDAP servers, slurpd runs on the master server and handles replication by pushing changes out to slave servers. A selection of command-line tools enable the querying and modification of LDAP data.

Setup is straightforward. First, create an /etc/openldap/slapd.conf file by copying the sample installed with the Open-LDAP server. These are the lines you need to pay attention to:

```
database bdb
suffix "dc=mydomain, dc=com"
rootdn "cn=admin, dc=mydomain, dc=com"
rootpw {crypt}abJnggxhB/yWI
directory /var/lib/ldap
```

The database format defaults to Berkeley DB, which is fine for data that will live within the OpenLDAP system. You can use a variety of other back ends, including ad hoc methods such as scripts that create the data on the fly.

suffix is your “LDAP basename.” It’s the root of your portion of the LDAP namespace, similar in concept to your DNS domain name. In fact, this example illustrates the use of a DNS domain name as an LDAP basename, which is a common practice.

rootdn is your administrator’s name, and rootpw is the administrator’s hashed password. Note that the domain components leading up to the administrator’s name must also be specified. You can use slappasswd to generate the value for this field; just copy and paste its output into the file.

Because of the presence of this password hash, make sure that the slapd.conf file is owned by root and that its permissions are 600.

Edit /etc/openldap/ldap.conf to set the default server and basename for LDAP client requests. It’s pretty straightforward—just set the argument of the host entry to the hostname of your server and set the base to the same value as the suffix in the slapd.conf file. Make sure both lines are uncommented. Here’s an example from atrust.com:

```
BASE dc=atrust,dc=com
URI ldap://atlantic.atrust.com
```

At this point, you can start up slapd simply by running it with no arguments.
389 Directory Server: alternative open source LDAP server

Like OpenLDAP, the 389 Directory Server (port389.org) is an extension of the work done at the University of Michigan. However, it spent some years in the commercial world (at Netscape) before returning as an open source project.

There are several reasons to consider the 389 Directory Server as an alternative to OpenLDAP, but its superior documentation is one clear advantage. The 389 Directory Server comes with several professional grade administration and use guides, including detailed installation and deployment instructions.

A few other key features of the 389 Directory Server are

- Multimaster replication for fault tolerance and superior write performance
- Active Directory user and group synchronization
- A graphical console for all facets of user, group, and server management
- On-line, zero-downtime, LDAP-based update of schema, configuration, management, and in-tree Access Control Information (ACIs)

389 Directory Server has a much more active development community than does OpenLDAP. We generally recommend it over OpenLDAP for new installations.

From an administrative standpoint, the structure and operation of the two open source servers are strikingly similar. This fact is perhaps not too surprising since both packages were built on the same original code base.
LDAP Querying

To administer LDAP, you need to be able to see and manipulate the contents of the database. The phpLDAPadmin tool mentioned earlier is one of the nicer free tools for this purpose because it gives you an intuitive point-and-click interface. If phpLDAPadmin isn’t an option, `ldapsearch` (distributed with both OpenLDAP and 389 Directory Server) is an analogous command-line tool that produces output in LDIF format. `ldapsearch` is especially good for use in scripts and for debugging environments in which Active Directory is acting as the LDAP server.

The following example query uses `ldapsearch` to look up directory information for every user whose `cn` starts with “ned.” In this case, there’s only one result. The meanings of the various command-line flags are discussed below.

```
$ ldapsearch -h atlantic.atrust.com -p 389
   -x -D "cn=trent,cn=users,dc=boulder,dc=atrust,dc=com" -W
   -b "cn=users,dc=boulder,dc=atrust,dc=com" "cn=ned*"
```

Enter LDAP Password: <password>

```
# LDAPv3
# base <cn=users,dc=boulder,dc=atrust,dc=com> with scope sub
# filter: cn=ned*
# requesting: ALL
#
# ned, Users, boulder.atrust.com
dn: cn=ned,cn=Users,dc=boulder,dc=atrust,dc=com
objectClass: top
objectClass: person
objectClass: organizationalPerson
objectClass: user
cn: ned
sn: McClain
telephoneNumber: 303 555 4505
givenName: Ned
distinguishedName: cn=ned,cn=Users,dc=boulder,dc=atrust,dc=com
displayName: Ned McClain
memberOf: cn=Users,cn=Builtin,dc=boulder,dc=atrust,dc=com
memberOf: cn=Enterprise Admins,cn=Users,dc=boulder,dc=atrust,dc=com
name: ned
sAMAccountName: ned
userPrincipalName: ned@boulder.atrust.com
lastLogonTimestamp: 129086952498943974
mail: ned@atrust.com
```

`ldapsearch`’s `-h` and `-p` flags specify the host and port of the LDAP server you want to query, respectively.
You usually need to authenticate yourself to the LDAP server. In this case, the `-x` flag requests simple authentication (as opposed to SASL). The `-D` flag identifies the distinguished name of a user account that has the privileges needed to execute the query, and the `-W` flag makes `ldapsearch` prompt for the corresponding password.

The `-b` flag tells `ldapsearch` where in the LDAP hierarchy to start the search. This parameter is known as the `baseDN`; hence the `b`. By default, `ldapsearch` returns all matching entries below the `baseDN`. You can tweak this behavior with the `-s` flag.

The last argument is a “filter,” which is a description of what you’re searching for. It doesn’t require an option flag. This filter, `cn=ned*`, returns all LDAP entries that have a common name that starts with “ned”. The filter is quoted to protect the star from shell globbing.

To extract all entries below a given `baseDN`, just use `objectClass=*` as the search filter—or leave the filter out, since this is the default.

Any arguments that follow the filter select specific attributes to return. For example, if you added `mail givenName` to the command line above, `ldapsearch` would return only the values of matching attributes.
Conversion of passwd and group files to LDAP

If you are moving to LDAP and your existing user and group information is stored in flat files, you may want to migrate your existing data. RFC2307 defines the standard mapping from traditional UNIX data sets, such as the passwd and group files, into the LDAP namespace. It’s a useful reference document for sysadmins who want to use LDAP in a UNIX environment, at least in theory. In practice, the specifications are a lot easier for computers to read than for humans; you’re better off looking at examples.

Padl Software offers a free set of Perl scripts that migrate existing flat files or NIS maps to LDAP. It’s available from padl.com/OSS/MigrationTools.html, and the scripts are straightforward to run. They can be used as filters to generate LDIF, or they can be run against a live server to upload the data directly. For example, the migrate_group script converts this line from /etc/group:

csstaff:x:2033:evi,matthew,trent

to the following LDIF:

dn: cn=csstaff,ou=Group,dc=domainname,dc=com
cn: csstaff
objectClass: posixGroup
objectClass: top
userPassword: {crypt}x
gidNumber: 2033
memberuid: evi
memberuid: matthew
memberuid: trent
17.3 Using directory services for login

Once you have a directory service set up, complete the following configuration chores so your system can enter SSO paradise:

- If you’re planning to use Active Directory with Kerberos, configure Kerberos and join the system to the Active Directory domain.
- Configure `sssd` to communicate with the appropriate identity and authentication stores (LDAP, Active Directory, or Kerberos).
- Configure the name service switch, `nsswitch.conf`, to use `sssd` as a source of user, group, and password information.
- Configure PAM to service authentication requests through `sssd`.

Some software uses the traditional `getpwent` family of library routines to look up user information, whereas modern services often directly call the PAM authentication routines. Configure both PAM and `nsswitch.conf` to ensure a fully functional environment.

We walk through these procedures below.
Kerberos

Kerberos is a ticket-based authentication system that uses symmetric key cryptography. Its recent popularity has been driven primarily by Microsoft, which uses it as part of Active Directory and Windows authentication. For SSO purposes, we describe how to integrate with an Active Directory Kerberos environment on both Linux and FreeBSD. If you’re using an LDAP server other than Active Directory or if you want to authenticate against Active Directory through the LDAP path rather than the Kerberos path, you can skip to the discussion of sssd here.

See this page for general information about Kerberos.

Linux Kerberos configuration for AD integration

Sysadmins often want their Linux systems to be members of an Active Directory domain. In the past, the complexity of this configuration drove some of those sysadmins to drink. Fortunately, the debut of realmd has made this task much simpler. realmd acts as a configuration tool for both sssd and Kerberos.

⚠️ Before attempting to join an Active Directory domain, verify the following:

- **realmd** is installed on the Linux system you’re joining to the domain.
- **sssd** is installed (see below).
- **ntpd** is installed and running.
- You know the correct name of your AD domain.
- You have credentials for an AD account that is allowed to join systems to the domain. This action results in a Kerberos ticket-granting ticket (TGT) being issued to the system so that it can perform authentication operations going forward without access to an administrator’s password.

For example, if your AD domain name is ULSAH.COM and the AD account trent is allowed to join systems to the domain, you can use the following command to join your system to the domain:

```
$ sudo realm join --user=trent ULSAH.COM
```

You can then verify the result:
FreeBSD Kerberos configuration for AD integration

Kerberos is infamous for its complex configuration process, especially on the server side. Unfortunately, FreeBSD has no slick tool akin to Linux’s `realmd` that configures Kerberos and joins an Active Directory domain in one step. However, you need to set up only the client side of Kerberos. The configuration file is `/etc/krb5.conf`.

First, double-check that the system’s fully qualified domain name has been included in `/etc/hosts` and that NTP is configured and working. Then edit `krb5.conf` to add the realm as shown in the following example. Substitute the name of your site’s AD domain for ULSAH.COM.

```plaintext
$ realm list
ulsah.com

  type: kerberos
  realm-name: ULSAH.COM
  domain-name: ulsah.com
  configured: kerberos-member
  server-software: active-directory
  client-software: sssd
  required-package: sssd
  required-package: adcli
  required-package: samba-common
  login-formats: %u@ulsah.com
  login-policy: allow-real logins

[logging]
  default = FILE:/var/log/krb5.log
[libdefaults]
  clockskew = 300
  default_realm = ULSAH.COM
  kdc_timesync = 1
  ccache_type = 4
  forwardable = true
  proxiable = true
[realms]
  ULSAH.COM = {
    kdc = dc.ulsah.com
    admin_server = dc.ulsah.com
    default_domain = ULSAH
  }
[domain_realm]
  .ulsah.com = ULSAH.COM
  ulsah.com = ULSAH.COM
```

Several values are of interest in the example above. A 5-minute clock skew is allowed even though the time is set through NTP. This leeway allows the system to function even in the event of an NTP problem. The default realm is set to the AD domain, and the key distribution center
(or KDC) is configured as an AD domain controller. `krb5.log` might come in handy for debugging.

Request a ticket from the Active Directory controller by running the `kinit` command. Specify a valid domain user account. The “administrator” account is usually a good test, but any account will do. When prompted, type the domain password.

```
$ kinit administrator@ULSAH.COM
Password for administrator@ULSAH.COM: <password>
```

Use `klist` to show the Kerberos ticket:

```
$ klist
Ticket cache: FILE:/tmp/krb5cc_1000
Default principal: administrator@ULSAH.COM

Valid starting       Expires            Service principal
04/30/17 13:40:19    04/30/17 23:40:21    krbtgt/ULSAH.COM@ULSAH.COM
          renew until 05/01/17 13:40:19

Kerberos 4 ticket cache: /tmp/tkt1000
klist: You have no tickets cached
```

If a ticket is displayed, authentication was successful. In this case, the ticket is valid for 10 hours and can be renewed for 24 hours. You can use the `kdestroy` command to invalidate the ticket.

The last step is to join the system to the domain, as shown below. The administrator account used (in this case, trent) must have the appropriate privileges on the Active Directory server to join systems to the domain.

```
$ net ads join -U trent
Enter trent's password: <password>
Using short domain -- ULSAH
Joined 'example.ulsah.com' to domain 'ULSAH.COM'
```

See the man page for `krb5.conf` for additional configuration options.
sssd: the System Security Services Daemon

The UNIX and Linux road to SSO nirvana has been a rough one. Years ago, it was common to set up independent authentication for every service or application. This approach often resulted in a morass of separate configurations and undocumented dependencies that were impossible to manage over time. Users’ passwords would work with one application but not another, causing frustration for everyone.

Microsoft formerly published extensions (originally called “Services for UNIX,” then “Windows Security and Directory Services for UNIX,” and finally, “Identity Management for UNIX” in Windows Server 2012) that facilitated the housing of UNIX users and groups within Active Directory. Putting the authority for managing these attributes in a non-UNIX system was an unnatural fit, however. To the relief of many, Microsoft discontinued this feature as of Windows Server 2016.

These issues needed some kind of comprehensive solution, and that’s just what we got with sssd, the System Security Services Daemon. Available for both Linux and FreeBSD, sssd is a one-stop shop for user identity wrangling, authentication, and account mapping. It can also cache credentials off-line, which is useful for mobile devices. sssd supports authentication both through native LDAP and through Kerberos.

You configure sssd through the sssd.conf file. Here’s a basic example for an environment that uses Active Directory as the directory service:

```conf
ds
services = nss, pam
domains = ULSAH.COM

domain/ULSAH.COM
id_provider = ad
access_provider = ad
```

If you are using a non-AD LDAP server, your sssd.conf file might look more like this:

```conf
ds
services = nss, pam
domains = LDAP

domain/LDAP
id_provider = ldap
auth_provider = ldap
ldap_uri = ldap://ldap.ulsah.com
ldap_user_search_base = dc=ulsah,dc=com
tls_reqcert = demand
ldap_tls_cacert = /etc/pki/tls/certs/ca-bundle.crt
```
For obvious security reasons, **sssd** does not allow authentication over an unencrypted channel, so the use of LDAPS/TLS is required. Setting the `tls_reqcert` attribute to `demand` in the example above forces **sssd** to validate the server certificate as an additional check. **sssd** drops the connection if the certificate is found to be deficient.

Once **sssd** is up and running, you must tell the system to use it as the source for identity and authentication information. Configuring the name service switch and configuring PAM are the next steps in this process.
**nsswitch.conf: the name service switch**

The name service switch (NSS) was developed to ease selection among various configuration databases and name resolution mechanisms. All the configuration goes into the `/etc/nsswitch.conf` file.

The syntax is simple: for a given type of lookup, you simply list the sources in the order they should be consulted. The system’s local `passwd` and `group` files should always be consulted first (specified by `files`), but you can then punt to Active Directory or another directory service by way of `sssd` (specified by `sss`). These entries do the trick:

```plaintext
passwd: files sss
group: files sss
shadow: files sss
```

Once you’ve configured the `nsswitch.conf` file, you can test the configuration with the command `getent passwd`. This command prints the user accounts defined by all sources in `/etc/passwd` format:

```
$ getent passwd
root:x:0:0:root:/root:/bin/bash
daemon:x:1:1:daemon:/usr/sbin:/bin/sh
...  
bwhaley:x:10006:10018::/home/bwhaley:/bin/sh
guest:*:10001:10001:Guest:/home/ULSAH/guest:/bin/bash
ben:*:10002:10000:Ben Whaley:/home/ULSAH/ben:/bin/bash
krbtgt:*:10003:10000:krbtgt:/home/ULSAH/krbtgt:/bin/bash
```

The only way to distinguish local users from domain accounts is by user ID and by the path of the home directory, as seen in the last three entries above.
PAM: cooking spray or authentication wonder?

PAM stands for “pluggable authentication modules.” The PAM system relieves programmers of the chore of implementing direct connections to authentication systems and gives sysadmins flexible, modular control over the system’s authentication methods. Both the concept and the term come from Sun Microsystems (now part of Oracle *sniff*) and from a 1996 paper by Samar and Lai of SunSoft.

In the distant past, commands like `login` included hardwired authentication code that prompted the user for a password, tested the password against the encrypted version obtained from `/etc/shadow` (*`/etc/passwd` at that time*), and rendered a judgment as to whether the two passwords matched. Of course, other commands (e.g., `passwd`) contained similar code. It was impossible to change authentication methods without source code, and administrators had little or no control over details such as whether the system should accept “password” as a valid password. PAM changed all that.

PAM puts the system’s authentication routines into a shared library that `login` and other programs can call. By separating authentication functions into a discrete subsystem, PAM makes it easy to integrate new advances in authentication and encryption into the computing environment. For instance, multifactor authentication is supported without changes to the source code of `login` and `passwd`.

For the sysadmin, setting the right level of security for authentication has become a simple configuration task. Programmers win, too, since they no longer have to write tedious authentication code. More importantly, their authentication systems are implemented correctly on the first try. PAM can authenticate all sorts of activities: user logins, other forms of system access, use of protected web sites, even the configuration of applications.

**PAM configuration**

PAM configuration files are a series of one-liners, each of which names a particular PAM module to be used on the system. The general format is

```
module-type control-flag module-path [ arguments ]
```

Fields are separated by whitespace.

The order in which modules appear in the PAM configuration file is important. For example, the module that prompts the user for a password must come before the module that checks that password for validity. One module can pass its output to the next by setting either environment variables or PAM variables.

The `module-type` parameter—auth, account, session, or password—determines what the module is expected to do. auth modules identify the user and grant group memberships. Modules that do account chores enforce restrictions such as limiting logins to particular times of day, limiting the
number of simultaneous users, or limiting the ports on which logins can occur. (For example, you
would use an account-type module to restrict root logins to the console.) session chores include tasks
that are done before or after a user is granted access; for example, mounting the user’s home
directory. Finally, password modules change a user’s password or passphrase.

The control-flag specifies how the modules in the stack should interact to produce an ultimate
result for the stack. Table 17.2 shows the common values.

Table 17.2: PAM control flags

<table>
<thead>
<tr>
<th>Flag</th>
<th>Stop on failure?</th>
<th>Stop on success?</th>
<th>Comments</th>
</tr>
</thead>
<tbody>
<tr>
<td>include</td>
<td>–</td>
<td>–</td>
<td>Includes another file at this point in the stack</td>
</tr>
<tr>
<td>optional</td>
<td>No</td>
<td>No</td>
<td>Significant only if this is the lone module</td>
</tr>
<tr>
<td>required</td>
<td>No</td>
<td>No</td>
<td>Failure eventually causes the stack to fail</td>
</tr>
<tr>
<td>requisite</td>
<td>Yes</td>
<td>No</td>
<td>Same as required, but fails stack immediately</td>
</tr>
<tr>
<td>sufficient</td>
<td>No</td>
<td>Yes</td>
<td>The name is kind of a lie; see comments below</td>
</tr>
</tbody>
</table>

If PAM could simply return a failure code as soon as the first individual module in a stack failed,
the control-flags system would be simpler. Unfortunately, the system is designed so that most
modules get a chance to run regardless of their sibling modules’ success or failure, and this fact
causes some subtleties in the flow of control. (The intent is to prevent an attacker from learning
which module in the PAM stack caused the failure.)

required modules are required to succeed; a failure of any one of them guarantees that the stack as a
whole will eventually fail. However, the failure of a module that is marked required doesn’t
immediately stop execution of the stack. If you want that behavior, use the requisite control flag
instead of required.

The success of a sufficient module aborts the stack immediately. However, the ultimate result of the
stack isn’t guaranteed to be a success because sufficient modules can’t override the failure of earlier
required modules. If an earlier required module has already failed, a successful sufficient module aborts
the stack and returns failure as the overall result.

Before you modify your systems’ security settings, make sure you understand the system
thoroughly and that you double-check the particulars. (You won’t configure PAM every day.
How long will you remember which version is requisite and which is required?)

PAM example

An example /etc/pam.d/login file from a Linux system running sssd is reproduced below. We
expanded the included files to form a more coherent example.
The auth stack includes several modules. On the first line, the pam_nologin module checks for the existence of the /etc/nologin file. If it exists, the module aborts the login immediately unless the user is root. The pam_securetty module ensures that root can log in only on terminals listed in /etc/securetty. This line uses an alternative Linux syntax described in the pam.conf man page. In this case, the requested behavior is similar to that of the required control flag. pam_env sets environment variables from /etc/security/pam_env.conf, then pam_unix2 checks the user’s credentials by performing standard UNIX authentication. If the user doesn’t have a local UNIX account, pam_sss attempts authentication through sssd. If any of these modules fail, the auth stack returns an error.

The account stack includes only the pam_unix2 and pam_sss modules. In this context, they assess the validity of the account itself. The modules return an error if, for example, the account has expired or the password must be changed. In the latter case, the relevant module collects a new password from the user and passes it on to the password modules.

The pam_pwcheck line checks the strength of proposed new passwords by calling the cracklib library. It returns an error if a new password does not meet the requirements. However, it also allows empty passwords because of the nullok flag. The pam_unix2 and pam_sss lines update the actual password.

Finally, the session modules perform several housekeeping chores. pam_loginuid sets the kernel’s loginuid process attribute to the user’s UID. pam_limits reads resource usage limits from /etc/security/limits.conf and sets the corresponding process parameters that enforce them. pam_unix2 and pam_sss log the user’s access to the system, and pam_umask sets an initial file creation
mode. The `pam_lastlog` module displays the user’s last login time as a security check, and the `pam_mail` module prints a note if the user has new mail. Finally, `pam_ck_connector` notifies the `ConsoleKit` daemon (a system-wide daemon that manages login sessions) of the new login.

At the end of the process, the user has been successfully authenticated and PAM returns control to `login`. 
17.4 Alternative approaches

Although LDAP is currently the most popular method for centralizing user identity and authentication information within an organization, many other approaches have emerged over the decades. Two older options, NIS and rsync, are still in use in some isolated pockets.
NIS: the Network Information Service

NIS, released by Sun in the 1980s, was the first “prime time” administrative database. It was originally called the Sun Yellow Pages, but eventually had to be renamed for legal reasons. NIS commands still begin with the letters yp, so it’s hard to forget the original name. NIS was widely adopted and is still supported in both FreeBSD and Linux.

These days, however, NIS is an old gray mare. NIS should not be used for new deployments, and existing deployments should be migrated to a modern day alternative such as LDAP.
**rsync: transfer files securely**

**rsync**, written by Andrew Tridgell and Paul Mackerras, is a bit like a souped-up version of **scp** that is scrupulous about preserving links, modification times, and permissions. It is network efficient because it looks inside individual files and attempts to transmit only the differences between versions.

One quick-and-dirty approach to distributing files such as `/etc/passwd` and `/etc/group` is to set up a **cron** job to **rsync** them from a master server. Although this scheme is easy to set up and might be useful in a pinch, it requires that all changes be applied directly to the master, including user password changes.

As an example, the command

```
# rsync -gopt -e ssh /etc/passwd /etc/shadow lollipop:/etc
```

transfers the `/etc/passwd` and `/etc/shadow` files to the machine lollipop. The `-gopt` options preserve the permissions, ownerships, and modification times of the file. **rsync** uses **ssh** as the transport, and so the connection is encrypted. However, **sshd** on lollipop must be configured not to require a password if you want to run this command from a script. Of course, such a setup has significant security implications. Coder beware!

With the **--include** and **--exclude** flags you can specify a list of regular expressions to match against filenames, so you can set up a sophisticated set of transfer criteria. If the command line gets too unwieldy, you can read the patterns from separate files with the **--include-file** and **--exclude-file** options.

Configuration management tools such as Ansible are another common way to distribute files among systems. See **Chapter 23, Configuration Management**, for more details.
17.5 **Recommended Reading**

A good general introduction to LDAP is *LDAP for Rocket Scientists*, which covers LDAP architecture and protocol. Find it on-line at [zytrax.com/books/ldap](http://zytrax.com/books/ldap). Another good source of information is the LDAP-related RFCs, which are numerous and varied. As a group, they tend to convey an impression of great complexity, which is somewhat unrepresentative of average use. Table 17.3 list some of the most important of these RFCs.

**Table 17.3: Important LDAP-related RFCs**

<table>
<thead>
<tr>
<th>RFC</th>
<th>Title</th>
</tr>
</thead>
<tbody>
<tr>
<td>2307</td>
<td>An Approach for Using LDAP as a Network Information Service</td>
</tr>
<tr>
<td>2820</td>
<td>Access Control Requirements for LDAP</td>
</tr>
<tr>
<td>2849</td>
<td>LDAP Data Interchange Format (LDIF)—Technical Specification</td>
</tr>
<tr>
<td>3112</td>
<td>LDAP Authentication Password Schema</td>
</tr>
<tr>
<td>3672</td>
<td>Subentries in the Lightweight Directory Access Protocol (LDAP)</td>
</tr>
<tr>
<td>4511</td>
<td>LDAP: The Protocol</td>
</tr>
<tr>
<td>4512</td>
<td>LDAP: Directory Information Models</td>
</tr>
<tr>
<td>4513</td>
<td>LDAP: Authentication Methods and Security Mechanisms</td>
</tr>
<tr>
<td>4514</td>
<td>LDAP: String Representation of Distinguished Names</td>
</tr>
<tr>
<td>4515</td>
<td>LDAP: String Representation of Search Filters</td>
</tr>
<tr>
<td>4516</td>
<td>LDAP: Uniform Resource Locator</td>
</tr>
<tr>
<td>4517</td>
<td>LDAP: Syntaxes and Matching Rules</td>
</tr>
<tr>
<td>4519</td>
<td>LDAP: Schema for User Applications</td>
</tr>
</tbody>
</table>

In addition, there are a couple of oldie-but-goodie books on LDAP:


There’s also a decent book focused entirely on PAM:


Finally, the O’Reilly book on Active Directory is excellent:

Decades ago, cooking a chicken dinner involved not just frying the chicken, but selecting a tender young chicken out of the coop, terminating it with a kill signal, plucking the feathers, etc. Today, most of us just buy a package of chicken at the grocery store or butcher shop and skip the mess.

Email has evolved in a similar way. Ages ago, it was common for organizations to hand-craft their email infrastructure, sometimes to the point of predetermining exact mail routing. Today, many organizations use packaged, cloud-hosted email services such as Google Gmail or Microsoft Office 365.

Even if your email system runs in the cloud, you will still have occasion to understand, support, and interact with it as an administrator. If your site uses local email servers, the workload expands even further to include configuration, monitoring, and testing chores.

If you find yourself in one of these more hands-on scenarios, this chapter is for you. Otherwise, skip this material and spend your email administration time responding to messages from wealthy foreigners who need help moving millions of dollars in exchange for a large reward. (Just kidding, of course.)
18.1 **Mail system architecture**

A mail system consists of several distinct components:

- A “mail user agent” (MUA or UA) that lets users read and compose mail
- A “mail submission agent” (MSA) that accepts outgoing mail from an MUA, grooms it, and submits it to the transport system
- A “mail transport agent” (MTA) that routes messages among machines
- A “delivery agent” (DA) that places messages in a local message store (the receiving users’ mailboxes or, sometimes, a database)
- An optional “access agent” (AA) that connects the user agent to the message store (e.g., through the IMAP or POP protocol)

Note that these functional divisions are somewhat abstract. Real-world mail systems break out these roles into somewhat different packages.

Attached to some of these functions are tools for recognizing spam, viruses, and (outbound) internal company secrets. Exhibit A illustrates how the various pieces fit together as a message winds its way from sender to receiver.

---

**Exhibit A: Mail system components**

<table>
<thead>
<tr>
<th>Host A – sender</th>
<th>Host B – receiver</th>
</tr>
</thead>
<tbody>
<tr>
<td><strong>UA</strong> Thunderbird</td>
<td><strong>MTA</strong> sendmail Exim MS Exchange Postfix (port 25)</td>
</tr>
<tr>
<td><strong>UA</strong> MS Outlook</td>
<td><strong>MTA</strong> sendmail Exim MS Exchange Postfix (port 25)</td>
</tr>
<tr>
<td><strong>UA</strong> macOS Mail</td>
<td><strong>DA</strong> local mail.local (port 25)</td>
</tr>
<tr>
<td><strong>UA</strong> Alpine</td>
<td><strong>MSA</strong> (port 587)</td>
</tr>
<tr>
<td><strong>UA</strong> /bin/mail</td>
<td><strong>AA</strong> Cyrus</td>
</tr>
</tbody>
</table>

**Legend:**
- UA = User agent
- MSA = Submission agent
- MTA = Transport agent
- DA = Delivery agent
- AA = Access agent

*to local user agents*
User agents

Email users run a user agent (sometimes called an email client) to read and compose messages. Email messages originally consisted only of text, but a standard known as Multipurpose Internet Mail Extensions (MIME) now encodes text formats and attachments (including viruses) into email. It is supported by most user agents. Since MIME generally does not affect the addressing or transport of mail, we do not discuss it further.

/bin/mail was the original user agent, and it remains the “good ol’ standby” for reading text email messages at a shell prompt. Since email on the Internet has moved far beyond the text era, text-based user agents are no longer practical for most users. But we shouldn’t throw /bin/mail away; it’s still a handy interface for scripts and other programs.

One of the elegant features illustrated in Exhibit A is that a user agent doesn’t necessarily need to be running on the same system—or even on the same platform—as the rest of your mail system. Users can reach their email from a Windows laptop or smartphone through access agent protocols such as IMAP and POP.
Submission agents

MSAs, a late addition to the email pantheon, were invented to offload some of the computational tasks of MTAs. MSAs make it easy for mail hub servers to distinguish incoming from outbound email (when making decisions about allowing relaying, for example) and give user agents a uniform and simple configuration for outbound mail.

The MSA is a sort of “receptionist” for new messages being injected into the system by local user agents. An MSA sits between the user agent and the transport agent and takes over several functions that were formerly a part of the MTA’s job. An MSA implements secure (encrypted and authenticated) communication with user agents and often does minor header rewriting and cleanup on incoming messages. In many cases, the MSA is really just the MTA listening on a different port with a different configuration applied.

MSAs speak the same mail transfer protocol used by MTAs, so they appear to be MTAs from the perspective of user agents. However, they typically listen for connections on port 587 rather than port 25, the MTA standard. For this scheme to work, user agents must connect on port 587 instead of port 25. If your user agents cannot be taught to use port 587, you can still run an MSA on port 25, but you must do so on a system other than the one that runs your MTA; only one process at a time can listen on a particular port.

If you use an MSA, be sure to configure your transport agent so that it doesn’t duplicate any of the rewriting or header fix-up work done by the MSA. Duplicate processing won’t affect the correctness of mail handling, but it does represent useless extra work.

See this page for more information about SMTP authentication.

Since your MSA uses your MTA to relay messages, the MSA and MTA must use SMTP-AUTH to authenticate each other. Otherwise, you create a so-called open relay that spammers can exploit and that other sites will blacklist you for.
Transport agents

A transport agent must accept mail from a user agent or submission agent, understand the recipients’ addresses, and somehow get the mail to the correct hosts for delivery. Transport agents speak the Simple Mail Transport Protocol (SMTP), which was originally defined in RFC821 but has now been superseded and extended by RFC5321. The extended version is called ESMTP.

An MTA’s list of chores, as both a mail sender and receiver, includes

- Receiving email messages from remote mail servers
- Understanding the recipients’ addresses
- Rewriting addresses to a form understood by the delivery agent
- Forwarding the message to the next responsible mail server or passing it to a local delivery agent to be saved to a user’s mailbox

The bulk of the work involved in setting up a mail system relates to the configuration of the MTA. In this book, we cover three open source MTAs: sendmail, Exim, and Postfix.
Local delivery agents

A delivery agent, sometimes called a local delivery agent (LDA), accepts mail from a transport agent and delivers it to the appropriate recipients’ mailboxes on the local machine. As originally specified, email can be delivered to a person, to a mailing list, to a file, or even to a program. However, the last two types of recipients can weaken the security and safety of your system.

MTAs usually include a built-in local delivery agent for easy deliveries. procmail (procmail.org) and Maildrop (courier-mta.org/maildrop) are LDAs that can filter or sort mail before delivering it. Some access agents (AAs) also have built-in LDAs that do both delivery and local housekeeping chores.
Message stores

A message store is the final resting place of an email message once it has completed its journey across the Internet and been delivered to recipients.

Mail has traditionally been stored in either mbox format or Maildir format. The former stores all mail in a single file, typically /var/mail/username, with individual messages separated by a special From line. Maildir format stores each message in a separate file. A file for each message is more convenient but creates directories with many, many small files; some filesystems may not be amused.

Flat files in mbox or Maildir format are still widely used, but ISPs with thousands or millions of email clients have typically migrated to other technologies for their message stores, usually databases. Unfortunately, that means that message stores are becoming more opaque.
Access agents

Two protocols access message stores and download email messages to a local device (workstation, laptop, smartphone, etc.): Internet Message Access Protocol version 4 (IMAP4) and Post Office Protocol version 3 (POP3). Earlier versions of these protocols had security issues. Be sure to use a version (IMAPS or POP3S) that incorporates SSL encryption and hence does not transmit passwords in cleartext over the Internet.

IMAP is significantly better than POP. It delivers your mail one message at a time rather than all at once, which is kinder to the network (especially on slow links) and better for someone who travels from location to location. IMAP is especially good at dealing with the giant attachments that some folks like to send: you can browse the headers of your messages and not download the attachments until you are ready to deal with them.
18.2 Anatomy of a Mail Message

A mail message has three distinct parts:

- Envelope
- Headers
- Body of the message

The envelope determines where the message will be delivered or, if the message can’t be delivered, to whom it should be returned. The envelope is invisible to users and is not part of the message itself; it’s used internally by the MTA.

Envelope addresses generally agree with the From and To lines of the header when the sender and recipient are individuals. The envelope and headers might not agree if the message was sent to a mailing list or was generated by a spammer who is trying to conceal his identity.

Headers are a collection of property/value pairs as specified in RFC5322 (updated by RFC6854). They record all kinds of information about the message, such as the date and time it was sent, the transport agents through which it passed on its journey, and who it is to and from. The headers are a bona fide part of the mail message, but user agents typically hide the less interesting ones when displaying messages for the user.

The body of the message is the content to be sent. It usually consists of plain text, although that text often represents a mail-safe encoding for various types of binary or rich-text content.

Dissecting mail headers to locate problems within the mail system is an essential sysadmin skill. Many user agents hide the headers, but there is usually a way to see them, even if you have to use an editor on the message store.

Below are most of the headers (with occasional truncations indicated by …) from a typical nonspam message. We removed another half page of headers that Gmail uses as part of its spam filtering. (In memory of Evi, who originally owned this chapter, this historical example has been kept intact.)
Delivered-To: sailingevi@gmail.com
Received: by 10.231.39.205 with SMTP id...; Fri, 24 May 2013 08:14:27
-700 (PDT)
Received: by 10.114.163.26 with SMTP id...; Fri, 24 May 2013 08:14:26
-700 (PDT)
Return-Path: <david@schweikert.ch>
Received: from mail-relay.atrust.com
(mail-relay.atrust.com [63.173.189.2]) by mx.google.com with
ESMTP id 17si2166978pxi.34.2009.10.16.08.14.20; Fri, 24 May 2013
08:14:25 -0700 (PDT)
Received-SPF: fail (google.com: domain of david@schweikert.
ch does not designate 63.173.189.2 as permitted sender)
Authentication-Results: mx.google.com; spf=hardfail (google.com: domain
of david@schweikert.ch does not designate 63.173.189.2 as permitted
sender) smtp.mail=david@schweikert.ch
Received: from mail.schweikert.ch (nigel.schweikert.ch [88.198.52.145])
by mail-relay.atrust.com (8.12.11/8.12.11) with ESMTP id n9GFEDKA0
for <evi@atrust.com>; Fri, 24 May 2013 09:14:14 -0600
Received: from localhost ([localhost.localdomain [127.0.0.1]]) by mail.
schweikert.ch (Postfix) with ESMTP id 3251112DA79; Fri, 24 May 2013
17:14:12 +0200 (CEST)
X-Virus-Scanned: Debian amavisd-new at mail.schweikert.ch
Received: from mail.schweikert.ch ([127.0.0.1]) by localhost ([mail.
schweikert.ch [127.0.0.1]) (amavisd-new, port 10024) with ESMTP id
dV8BpT7rhJkC; Fri, 24 May 2013 17:14:07 +0200 (CEST)...
Received: by mail.schweikert.ch (Postfix, from user id 1000)
  id 2A156120B89; Fri, 24 May 2013 17:14:07 +0200 (CEST)
Date: Fri, 24 May 2013 17:14:06 +0200
From: David Schweikert <david@schweikert.ch>
To: evi@atrust.com
Cc: Garth Snyder <garth@garthsnyder.com>
Subject: Email chapter comments

Hi evi,

I just finished reading the email chapter draft, and I was pleased to see ...

To decode this beast, start reading the Received lines, but start from the bottom (sender side). This message went from David Schweikert’s home machine in the schweikert.ch domain to his mail server (mail.schweikert.ch), where it was scanned for viruses. It was then forwarded to the recipient evi@atrust.com. However, the receiving host mail-relay.atrust.com sent it on to sailingevi@gmail.com, where it entered Evi’s mailbox.

See this page for more information about SPF.
Midway through the headers, you see an SPF (Sender Policy Framework) validation failure, an indication that the message has been flagged as spam. This failure happened because Google checked the IP address of mail-relay.atrust.com and compared it with the SPF record at schweikert.ch; of course, it doesn’t match. This is an inherent weakness of relying on SPF records to identify forgeries—they don’t work for mail that has been relayed.

You can often see the MTAs that were used (Postfix at schweikert.ch, sendmail 8.12 at atrust.com), and in this case, you can also see that virus scanning was performed through amavisd-new on port 10,024 on a machine running Debian Linux. You can follow the progress of the message from the Central European Summer Time zone (CEST +0200), to Colorado (-0600), and on to the Gmail server (PDT -0700); the numbers are the differences between local time and UTC, Coordinated Universal Time. A lot of info is stashed in the headers!

Here are the headers, again truncated, from a spam message:

```
Delivered-To: sailingevi@gmail.com
Received: by 10.231.39.205 with SMTP id...; Fri, 19 Oct 2009 08:59:32
   -0700...
Received: by 10.231.5.143 with SMTP id...; Fri, 19 Oct 2009 08:59:31
   -0700...
Return-Path: <smotheringl39@sherman.dp.ua>
Received: from mail-relay.atrust.com (mail-relay.atrust.com
   [63.173.189.2]) ... 
Received-SPF: neutral (google.com: 63.173.189.2 is neither
   permitted nor denied by best guess record for domain of
   smotheringl39@sherman.dp.ua) client-ip=63.173.189.2;
Authentication-Results: mx.google.com; spf=neutral (google. 
   com: 63.173.189.2 is neither permitted nor denied by best
   guess record for domain of smotheringl39@sherman.dp.ua)
   smtp.mail=smotheringl39@sherman.dp.ua
Received: from SpeedTouch.lan (187-10-167-249.dsl.telesp.net.br
   [187.10.167.249] (may be forged)) by mail-relay.atrust.com ...
Received: from 187.10.167.249 by relay2.trifle.net; Fri, 19 Oct 2009
   13:59: ... 
From: "alert@atrust.com" <alert@atrust.com>
To: <ned@atrust.com>
Subject: A new settings file for the ned@atrust.com mailbox
Date: Fri, 19 Oct 2009 13:59:12 -0300 ...
```

According to the From header, this message’s sender is alert@atrust.com. But according to the Return-Path header, which contains a copy of the envelope sender, the originator was smotheringl39@sherman.dp.ua, an address in the Ukraine. The first MTA that handled the message is at IP address 187.10.167.249, which is in Brazil. Sneaky spammers... It’s important to note that many of the lines in the header, including the Received lines, may have been forged. Use this data with extreme caution.
The SPF check at Google fails again, this time with a “neutral” result because the domain sherman.dp.ua does not have an SPF record with which to compare the IP address of mail-relay.atrust.com.

The recipient information is also at least partially untrue. The To header says the message is addressed to ned@atrust.com. However, the envelope recipient addresses must have included evi@atrust.com in order for the message to be forwarded to sailingevi@gmail.com for delivery.
18.3 The SMTP Protocol

The Simple Mail Transport Protocol (SMTP) and its extended version, ESMTP, have been standardized in the RFC series (RFC5321, updated by RFC7504) and are used for most message hand-offs among the various pieces of the mail system:

- UA-to-MSA or -MTA as a message is injected into the mail system
- MSA-to-MTA as the message starts its delivery journey
- MTA- or MSA-to-antivirus or -antispam scanning programs
- MTA-to-MTA as a message is forwarded from one site to another
- MTA-to-DA as a message is delivered to the local message store

Because the format of messages and the transfer protocol are both standardized, my MTA and your MTA don’t have to be the same or even know each other’s identity; they just have to both speak SMTP or ESMTP. Your various mail servers can run different MTAs and interoperate just fine.

True to its name, SMTP is…simple. An MTA connects to your mail server and says, “Here’s a message; please deliver it to user@your.domain.” Your MTA says “OK.”

Requiring strict adherence to the SMTP protocol has become a technique for fighting spam and malware, so it’s important for mail administrators to be somewhat familiar with the protocol. The language has only a few commands; Table 18.1 shows the most important ones.

Table 18.1: SMTP commands

<table>
<thead>
<tr>
<th>Command</th>
<th>Function</th>
</tr>
</thead>
<tbody>
<tr>
<td>HELO hostname</td>
<td>Identifies the connecting host if speaking SMTP</td>
</tr>
<tr>
<td>EHLO hostname</td>
<td>Identifies the connecting host if speaking ESMTP</td>
</tr>
<tr>
<td>MAIL FROM: revpath</td>
<td>Initiates a mail transaction (envelope sender)</td>
</tr>
<tr>
<td>RCPT TO: fwdpatha</td>
<td>Identifies envelope recipient(s)</td>
</tr>
<tr>
<td>VRFY address</td>
<td>Verifies that address is valid (deliverable)</td>
</tr>
<tr>
<td>EXPN address</td>
<td>Shows expansion of aliases and .forward mappings</td>
</tr>
<tr>
<td>DATA</td>
<td>Begins the message body (preceded by headers)b</td>
</tr>
<tr>
<td>QUIT</td>
<td>Ends the exchange and closes the connection</td>
</tr>
<tr>
<td>RSET</td>
<td>Resets the state of the connection</td>
</tr>
<tr>
<td>HELP</td>
<td>Prints a summary of SMTP commands</td>
</tr>
</tbody>
</table>

a. There can be multiple RCPT commands for a message.
b. You terminate the body by entering a dot on its own line.
**You had me at EHLO**

ESMTP speakers start conversations with EHLO instead of HELO. If the process at the other end understands and responds with an OK, then the participants negotiate supported extensions and agree on a lowest common denominator for the exchange. If the peer returns an error in response to the EHLO, then the ESMTP speaker falls back to SMTP. But today, almost everything uses ESMTP.

A typical SMTP conversation to deliver an email message goes as follows: HELO or EHLO, MAIL FROM:, RCPT TO:, DATA, and QUIT. The sender does most of the talking, with the recipient contributing error codes and acknowledgments.

SMTP and ESMTP are both text-based protocols, so you can use them directly when debugging the mail system. Just **telnet** to TCP port 25 or 587 and start entering SMTP commands. See the example [here](#).
SMTP error codes

Also specified in the RFCs that define SMTP are a set of temporary and permanent error codes. These were originally three-digit codes (e.g., 550), with each digit being interpreted separately. A first digit of 2 indicated success, a 4 signified a temporary error, and a 5 indicated a permanent error.

The three-digit error code system did not scale, so RFC3463 (updated by RFCs 3886, 4468, 4865, 4954, and 5248) restructured it to create more flexibility. It defined an expanded error code format known as a delivery status notification or DSN. DSNs have the format X.X.X instead of the old XXX, and each of the individual Xs can be a multidigit number. The initial X must still be 2, 4, or 5. The second digit specifies a topic, and the third provides the details. The new system uses the second number to distinguish host errors from mailbox errors. Table 18.2 lists a few of the DSN codes. RFC3463’s Appendix A shows them all.

Table 18.2: RFC3463 delivery status notifications

<table>
<thead>
<tr>
<th>Temporary</th>
<th>Permanent</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>4.2.1</td>
<td>5.2.1</td>
<td>Mailbox is disabled</td>
</tr>
<tr>
<td>4.2.2</td>
<td>5.2.2</td>
<td>Mailbox is full</td>
</tr>
<tr>
<td>4.2.3</td>
<td>5.2.3</td>
<td>Message is too long</td>
</tr>
<tr>
<td>4.4.1</td>
<td>5.4.1</td>
<td>No answer from host</td>
</tr>
<tr>
<td>4.4.4</td>
<td>5.4.4</td>
<td>Unable to route</td>
</tr>
<tr>
<td>4.5.3</td>
<td>5.5.3</td>
<td>Too many recipients</td>
</tr>
<tr>
<td>4.7.1</td>
<td>5.7.1</td>
<td>Delivery not authorized, message refused</td>
</tr>
<tr>
<td>4.7.*</td>
<td>5.7.*</td>
<td>Site policy violation</td>
</tr>
</tbody>
</table>
SMTP authentication

RFC4954 (updated by RFC5248) defines an extension to the original SMTP protocol that allows an SMTP client to identify and authenticate itself to a mail server. The server might then let the client relay mail through it. The protocol supports several different authentication mechanisms. The exchange is as follows:

1. The client says EHLO, announcing that it speaks ESMTP.
2. The server responds and advertises its authentication mechanisms.
3. The client says AUTH and names a specific mechanism that it wants to use, optionally including its authentication data.
4. The server accepts the data sent with AUTH or starts a challenge and response sequence with the client.
5. The server either accepts or denies the authentication attempt.

To see what authentication mechanisms a server supports, you can telnet to port 25 and say EHLO. For example, here is a truncated conversation with the mail server mail-relay.atrust.com (the commands we typed are in bold):

```
$ telnet mail-relay.atrust.com 25
Trying 192.168.2.1...
Connected to mail-relay.atrust.com.
Escape character is '^]'.
220 mail-relay.atrust.com ESMTP AT Mail Service 28.1.2/28.1.2; Mon, 12 Sep 2016 18:05:55 -0600
ehlo booklab.atrust.com
250-mail-relay.atrust.com Hello [192.168.22.35], pleased to meet you
250-ENHANCEDSTATUSCODES
250-PIPELINING
250-8BITMIME
250-SIZE
250-DSN
250-ETRN
250-AUTH LOGIN PLAIN
250-DELIVERBY
250 HELP
```

In this case, the mail server supports the LOGIN and PLAIN authentication mechanisms. sendmail, Exim, and Postfix all support SMTP authentication; details of configuration are covered here, here, and here, respectively.
18.4 Spam and Malware

Spam is the jargon word for junk mail, also known as unsolicited commercial email or UCE. It is one of the most universally hated aspects of the Internet. Once upon a time, system administrators spent many hours each week hand-tuning block lists and adjusting decision weights in home-grown spam filtering tools. Unfortunately, spammers have become so crafty and commercialized that these measures are no longer an effective use of system administrators’ time.

In this section we cover the basic antispam features of each MTA. However, there’s a certain futility to any attempt to fight spam as a lone vigilante. You should really pay for a cloud-based spam-fighting service (such as McAfee SaaS Email Protection, Google G Suite, or Barracuda) and leave the spam fighting to the professionals who love that stuff. They have better intelligence about the state of the global emailsphere and can react far more quickly to new information than you can.

Spam has become a serious problem because although the absolute response rate is low, the responses per dollar spent is high. (A list of 30 million email addresses costs about $20.) If it didn’t work for the spammers, it wouldn’t be such a problem. Surveys show that 95%–98% of all mail is spam.

There are even venture-capital-funded companies whose entire mission is to deliver spam less expensively and more efficiently (although they typically call it “marketing email” rather than spam). If you work at or buy services from one of these companies, we’re not sure how you sleep at night.

In all cases, advise your users to simply delete the spam they receive. Many spam messages contain instructions that purport to explain how recipients can be removed from the mailing list. If you follow those instructions, however, the spammers may remove you from the current list, but they immediately add you to several other lists with the annotation “reaches a real human who reads the message.” Your email address is then worth even more.
Forgeries

Forging email is trivial; many user agents let you fill in the sender’s address with anything you want. MTAs can use SMTP authentication between local servers, but that doesn’t scale to Internet sizes. Some MTAs add warning headers to outgoing local messages that they think might be forged.

Any user can be impersonated in mail messages. Be careful if email is your organization’s authorization vehicle for things like door keys, access cards, and money. The practice of targeting users with forged email is commonly called “phishing.” You should warn administrative users of this fact and suggest that if they see suspicious mail that appears to come from a person in authority, they should verify the validity of the message. Caution is doubly appropriate if the message asks that unreasonable privileges be given to an unusual person.
SPF and Sender ID

The best way to fight spam is to stop it at its source. This sounds simple and easy, but in reality it’s almost an impossible challenge. The structure of the Internet makes it difficult to track the real source of a message and to verify its authenticity. The community needs a sure-fire way to verify that the entity sending an email is actually who or what it claims to be. Many proposals have addressed this problem, but SPF and Sender ID have achieved the most traction.

SPF, or Sender Policy Framework, has been described by the IETF in RFC7208. SPF defines a set of DNS records through which an organization can identify its official outbound mail servers. MTAs can then refuse email purporting to be from that organization’s domain if the email does not originate from one of these official sources. Of course, the system only works well if the majority of organizations publish SPF records.

Sender ID and SPF are virtually identical in form and function. However, key parts of Sender ID are patented by Microsoft, and hence it has been the subject of much controversy. As of this writing (2017), Microsoft is still trying to strong-arm the industry into adopting its proprietary standards. The IETF chose not to choose and published RFC4406 on Sender ID and RFC7208 on SPF. Organizations that implement this type of spam avoidance strategy typically use SPF.

Messages that are relayed break both SPF and Sender ID, which is a serious flaw in these systems. The receiver consults the SPF record for the original sender to discover its list of authorized servers. However, those addresses won’t match any relay machines that were involved in transporting the message. Be careful what decisions you make in response to SPF failures.
DKIM

DKIM (DomainKeys Identified Mail) is a cryptographic signature system for email messages. It lets the receiver verify not only the sender’s identity but also the fact that a message has not been tampered with in transit. The system uses DNS records to publish a domain’s cryptographic keys and message-signing policy. DKIM is supported by all the MTAs described in this chapter, but real-world deployment has been extremely rare.
18.5 Message privacy and encryption

By default, all mail is sent unencrypted. Educate your users that they should never send sensitive data through email unless they make use of an external encryption package or your organization has provided a centralized encryption solution for email. Even with encryption, electronic communication can never be guaranteed to be 100% secure. You pays your money and you takes your chances.

Historically, the most common external encryption packages have been Pretty Good Privacy (PGP), its GNUified clone GPG, and S/MIME. Both S/MIME and PGP are documented in the RFC series, with S/MIME being on the standards track. Most common user agents support plugins for both solutions.

These standards offer a basis for email confidentiality, authentication, message integrity assurance, and nonrepudiation of origin. But although PGP/GPG and S/MIME are potentially viable solutions for tech-savvy users who care about privacy, they have proved too cumbersome for unsophisticated users. Both require some facility with cryptographic key management and an understanding of the underlying encryption strategy. (Pro tip: If you use PGP/GPG or S/MIME, you can increase your odds of remaining secure by ensuring that your public key or certificate is expired and replaced frequently. Long-term use of a key increases the likelihood that it will be compromised without your awareness.)

Most organizations that handle sensitive data in email (especially ones that communicate with the public, such as health care institutions) opt for a centralized service that uses proprietary technology to encrypt messages. Such systems can use either on-premises solutions (such as Cisco’s IronPort) that you deploy in your data center or cloud-based services (such as Zix, zixcorp.com) that can be configured to encrypt outbound messages according to their contents or other rules. Centralized email encryption is one category of service for which it’s best to use a commercial solution rather than rolling your own.

At least in the email realm, data loss prevention (DLP) is a kissing cousin to centralized encryption. DLP systems seek to avoid—or at least, detect—the leakage of proprietary information into the stream of email leaving your organization. They scan outbound email for potentially sensitive content. Suspicious messages can be flagged, blocked, or returned to their senders. Our recommendation is that you choose a centralized encryption platform that also includes DLP capability; it’s one less platform to manage.

See this page for more information about TLS.

In addition to encrypting transport between MTAs, it’s important to ensure that user-agent-to-access-agent communication is always encrypted, especially because this channel typically employs some form of user credentials to connect. Make sure that only the secure, TLS-using
versions of the IMAP and POP protocols are allowed by access agents. (These are known as IMAPS and POP3S, respectively.)
18.6 Mail aliases

Another concept that is common to all MTAs is the use of aliases. Aliases allow mail to be rerouted either by the system administrator or by individual users.

Aliases can define mailing lists, forward mail among machines, or allow users to be referred to by more than one name. Alias processing is recursive, so it’s legal for an alias to point to other destinations that are themselves aliases.

Technically, aliases are configured only by sysadmins. A user’s control of mail routing through the use of a .forward file is not really aliasing, but we have lumped them together here.

Sysadmins often use role or functional aliases (e.g., printers@example.com) to route email about a particular issue to whatever person is currently handling that issue. Other examples might include an alias that receives the results of a nightly security scan or an alias for the postmaster in charge of email.

The most common method for configuring aliases is to use a simple flat file such as the /etc/mail/aliases file discussed later in this section. This method was originally introduced by sendmail, but Exim and Postfix support it, too.

Most user agents also provide some sort of “aliasing” feature (usually called “my groups,” “my mailing lists,” or something of that nature). However, the user agent expands such aliases before mail ever reaches an MSA or MTA. These aliases are internal to the user agent and don’t require support from the rest of the mail system.

Aliases can also be defined in a forwarding file in the home directory of each user, usually ~/.forward. These aliases, which use a slightly nonstandard syntax, apply to all mail delivered to that particular user. They’re often used to forward mail to a different account or to implement automatic “I’m on vacation” responses.

MTAs look for aliases in the global aliases file (/etc/mail/aliases or /etc/aliases) and then in recipients’ forwarding files. Aliasing is applied only to messages that the transport agent considers to be local.

The format of an entry in the aliases file is

\[ \text{local-name: recipient1,recipient2,...} \]

where local-name is the original address to be matched against incoming messages and the recipient list contains either recipient addresses or the names of other aliases. Indented lines are considered continuations of the preceding lines.

From mail’s point of view, the aliases file supersedes /etc/passwd, so the entry
would prevent the local user david from ever receiving any mail. Therefore, administrators and `adduser` tools should check both the `passwd` file and the `aliases` file when selecting new usernames.

The `aliases` file should always contain an alias named “postmaster” that forwards mail to whoever maintains the mail system. Similarly, an alias for “abuse” is appropriate in case someone outside your organization needs to contact you regarding spam or suspicious network behavior that originates at your site. An alias for automatic messages from the MTA must also be present; it’s usually called Mailer-Daemon and is often aliased to postmaster.

Sadly, the mail system is so commonly abused these days that some sites configure their standard contact addresses to throw mail away instead of forwarding it to a human user. Entries such as

```bash
# Basic system aliases -- these MUST be present
mailer-daemon: postmaster
postmaster:  
'''
```

are common. We don’t recommend this practice, because humans who are having trouble reaching your site by email do sometimes write to the postmaster address.

A better paradigm might be

```bash
# Basic system aliases -- these MUST be present
mailer-daemon: 
postmaster:  
```

You should redirect root’s mail to your site’s sysadmins or to someone who logs in every day. The bin, sys, daemon, nobody, and hostmaster accounts (and any other site-specific pseudo-user accounts you set up) should all have similar aliases.

In addition to a list of users, aliases can refer to

- A file containing a list of addresses
- A file to which messages should be appended
- A command to which messages should be given as input

These last two targets should push your “What about security?” button, because the sender of a message totally determines its content. Being able to append that content to a file or deliver it as input to a command sounds pretty scary. Many MTAs either disallow these alias targets or severely limit the commands and file permissions that are acceptable.

Aliases can cause mail loops. MTAs try to detect loops that would cause mail to be forwarded back and forth forever and return the errant messages to the sender. To determine when mail is
looping, an MTA can count the number of Received lines in a message’s header and stop forwarding it when the count reaches a preset limit (usually 25). Each visit to a new machine is called a “hop” in email jargon; returning a message to the sender is known as “bouncing” it. So a more typically jargonized summary of loop handling would be, “Mail bounces after 25 hops.”

Another way MTAs can detect mail loops is by adding a Delivered-To header for each host to which a message is forwarded. If an MTA finds itself wanting to send a message to a host that’s already mentioned in a Delivered-To header, it knows the message has traveled in a loop.

In this chapter, we sometimes call a returned message a “bounce” and sometimes call it an “error.” What we really mean is that a delivery status notification (DSN, a specially formatted email message) has been generated. Such a notification usually means that a message was undeliverable and is therefore being returned to the sender.
Getting aliases from files

The :include: directive in the aliases file (or a user’s .forward file) allows the list of targets for the alias to be taken from the specified file. It is a great way to let users manage their own local mailing lists. The included file can be owned by the user and changed without involving a system administrator. However, such an alias can also become a tasty and effective spam expander, so don’t let email from outside your site be directed there.

When setting up a list to use :include:, the sysadmin must enter the alias into the global aliases file, create the included file, and chown the included file to the user that is maintaining the mailing list. For example, the aliases file might contain

sa-book: :include:/usr/local/mail/ulsah.authors

The file ulsah.authors should be on a local filesystem and should be writable only by its owner. To be complete, we should also include aliases for the mailing list’s owner so that errors (bounces) are sent to the owner of the list and not to the sender of a message addressed to the list:

owner-sa-book: evi
**Mailing to files**

If the target of an alias is an absolute pathname, messages are appended to the specified file. The file must already exist. For example:

```
cron-status: /usr/local/admin/cron-status-messages
```

If the pathname includes special characters, it must be enclosed in double quotes.

It’s useful to be able to send mail to files, but this feature arouses the interest of the security police and is therefore restricted. This syntax is only valid in the `aliases` file and in a user’s `.forward` file (or in a file that’s interpolated into one of these files with the `:include:` directive). A filename is not understood as a normal address, so mail addressed to `/etc/passwd@example.com` would bounce.

If the destination file is referenced from the `aliases` file, it must be world-writable (not advisable), setuid but not executable, or owned by the MTA’s default user. The identity of the default user is set in the MTA’s configuration file.

If the file is referenced in a `.forward` file, it must be owned and writable by the original message recipient, who must be a valid user with an entry in the `passwd` file and a valid shell that’s listed in `/etc/shells`. For files owned by root, use mode `4644` or `4600`, setuid but not executable.
Mailing to programs

An alias can also route mail to the standard input of a program. This behavior is specified with a line such as

```plaintext
autolog: "|/usr/local/bin/autologger"
```

It’s even easier to create security holes with this feature than with mailing to a file, so once again it is only permitted in `aliases`, `.forward`, or `::include:` files, and often requires the use of a restricted shell.
Building the hashed alias database

Since entries in the **aliases** file are unordered, it would be inefficient for the MTA to search this file directly. Instead, a hashed version is constructed with the Berkeley DB system. Hashing significantly speeds alias lookups, especially when the file gets large.

The file derived from `/etc/mail/aliases` is called **aliases.db**. If you are running Postfix or **sendmail**, you must rebuild the hashed database with the **newaliases** command every time you change the **aliases** file. Exim detects changes to the **aliases** file automatically. Save the error output if you run **newaliases** automatically—you might have introduced formatting errors in the **aliases** file.
18.7 Email configuration

The heart of an email system is its MTA, or mail transport agent. sendmail is the original UNIX MTA, written by Eric Allman while he was a graduate student many years ago. Since then, a host of other MTAs have been developed. Some of them are commercial products and some are open source implementations. In this chapter, we cover three open source mail-transport agents: sendmail, Postfix by Wietse Venema of IBM Research, and Exim by Philip Hazel of the University of Cambridge.

Configuration of the MTA can be a significant sysadmin chore. Fortunately, the default or sample configurations that ship with MTAs are often close to what the average site needs. You need not start from scratch when configuring your MTA.

SecuritySpace (securityspace.com) does a survey monthly to determine the market share of the various MTAs. In their June 2017 survey, 1.7 million out of 2 million MTAs surveyed replied with a banner that identified the MTA software in use. Table 18.3 shows these results, as well as the SecuritySpace results for 2009 and some 2001 values from a different survey.

Table 18.3: Mail transport agent market share

<table>
<thead>
<tr>
<th>MTA</th>
<th>Source</th>
<th>Default MTA on</th>
<th>Market share</th>
</tr>
</thead>
<tbody>
<tr>
<td></td>
<td></td>
<td></td>
<td>2017</td>
</tr>
<tr>
<td>Exim</td>
<td>exim.org</td>
<td>Debian</td>
<td>56%</td>
</tr>
<tr>
<td>Postfix</td>
<td>postfix.org</td>
<td>Red Hat, Ubuntu</td>
<td>33%</td>
</tr>
<tr>
<td>Exchange</td>
<td>microsoft.com/exchange</td>
<td>–</td>
<td>1%</td>
</tr>
<tr>
<td>sendmail</td>
<td>sendmail.org</td>
<td>FreeBSD</td>
<td>5%</td>
</tr>
<tr>
<td>All others</td>
<td>–</td>
<td>–</td>
<td>&lt;3% ea</td>
</tr>
</tbody>
</table>

The trend is clearly away from sendmail and toward Exim and Postfix, with Microsoft dropping to almost nothing. Keep in mind that this data includes only MTAs that are directly exposed to the Internet.

For each of the MTAs we cover, we include details on the common areas of interest:

- Configuration of simple clients
- Configuration of an Internet-facing mail server
- Control of both inbound and outbound mail routing
- Stamping of mail as coming from a central server or the domain itself
- Security
If you are implementing a mail system from scratch and have no site politics or biases to deal with, you may find it hard to choose an MTA. **sendmail** is largely out of vogue, with the possible exception of pure FreeBSD sites. Exim is powerful and highly configurable but suffers in complexity. Postfix is simpler, faster, and was designed with security as a primary goal. If your site or your sysadmins have a history with a particular MTA, it’s probably not worth switching unless you need features that are not available from your old MTA.

**sendmail** configuration is covered in the next section. Exim configuration begins [here](#), and Postfix configuration [here](#).
18.8 SENDMAIL

The sendmail distribution is available in source form from sendmail.org, but it’s rarely necessary to build sendmail from scratch these days. If you must do so, refer to the top-level INSTALL file for instructions. To tweak some of the build defaults, look up sendmail’s assumptions in devtools/OS/your-OS-name. Add features by editing devtools/Site/site.config.m4. As of October 2013, sendmail is supported and distributed by Proofpoint, Inc., a public company.

sendmail uses the m4 macro preprocessor not only for compilation but also for configuration. An m4 configuration file is usually named hostname.mc and is then translated from a slightly user-friendly syntax into a totally inscrutable low-level language in the file hostname.cf, which is in turn installed as /etc/mail/sendmail.cf.

To see what version of sendmail is installed on your system and how it was compiled, try the following command:

```
linux$ /usr/sbin/sendmail -d0.1 -bt < /dev/null
Version 8.13.8
Compiled with: DNSMAP HESIOD HES_GETMAILHOST LDAPMAP LOG
MAP_REX MATCHGECOS MILTER MIME7TO8 MIME8TO7 NAMED_BIND
NETINET NETINET6 NETUNIX NEWDB NIS PIPELINING SASLv2 SCANF
SOCKETMAP STARTTLS TCPRWRAPPERS USERDB USE_LDAP_INIT
======== SYSTEM IDENTITY (after readcf) ========
   (short domain name) $w = ross
   (canonical domain name) $j = ross.atrust.com
   (subdomain name) $m = atrust.com
   (node name) $k = ross.atrust.com
=====================================================================
```

This command puts sendmail in address test mode (-bt) and debug mode (-d0.1) but gives it no addresses to test (</dev/null). A side effect is that sendmail tells us its version and the compiler flags it was built with. Once you know the version number, you can look at the sendmail.org web site to see if any known security vulnerabilities are associated with that release.

To find the sendmail files on your system, look at the beginning of the installed /etc/mail/sendmail.cf file. The comments there mention the directory in which the configuration was built. That directory should in turn lead you to the .mc file that is the original source of the configuration.

Most vendors that ship sendmail include not only the binary but also the cf directory from the distribution tree, which they hide somewhere among the operating system files. Table 18.4 will help you find it.

Table 18.4: Config directory locations
<table>
<thead>
<tr>
<th>System</th>
<th>Directory</th>
</tr>
</thead>
<tbody>
<tr>
<td>Ubuntu</td>
<td>/usr/share/sendmail</td>
</tr>
<tr>
<td>Debian</td>
<td>/usr/share/sendmail</td>
</tr>
<tr>
<td>Red Hat</td>
<td>/etc/mail</td>
</tr>
<tr>
<td>CentOS</td>
<td>/etc/mail</td>
</tr>
<tr>
<td>FreeBSD</td>
<td>/etc/mail</td>
</tr>
</tbody>
</table>
The switch file

The service switch is covered in more detail starting here.

Most systems have a “service switch” configuration file, /etc/nsswitch.conf, that enumerates the methods that can satisfy various standard queries such as user and host lookups. If more than one resolution method is listed for a given type of query, the service switch file also determines the order in which the various methods are consulted.

The existence of the service switch is normally transparent to software. However, sendmail likes to exert fine-grained control over its lookups, so it currently ignores the system switch file and instead uses its own internal service configuration file (/etc/mail/service.switch).

Two fields in the switch file impact the mail system: aliases and hosts. The possible values for the hosts service are dns, nis, nisplus, and files. For aliases, the possible values are files, nis, nisplus, and ldap. Support for the mechanisms you use (except files) must be compiled into sendmail before the service can be used.
Starting sendmail

`sendmail` should not be controlled by `inetd` or `systemd`, so it must be explicitly started at boot time. See Chapter 2, *Booting and System Management Daemons*, for startup details.

The flags that `sendmail` is started with determine its behavior. You can run it in several different modes, selected with the `-b` flag. `-b` stands for “be” or “become” and is always used with another flag that determines the role `sendmail` will play. Table 18.5 lists the legal values and also includes the `-A` flag, which selects between MTA and MSA behavior.

Table 18.5: Command-line flags for `sendmail`’s major modes

<table>
<thead>
<tr>
<th>Flag</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td><code>-Ac</code></td>
<td>Uses the <code>submit.cf</code> config file and acts as an MSA</td>
</tr>
<tr>
<td><code>-Am</code></td>
<td>Uses the <code>sendmail.cf</code> config file and acts as an MTA</td>
</tr>
<tr>
<td><code>-ba</code></td>
<td>Runs in ARPANET mode (expects CR/LF at the ends of lines)</td>
</tr>
<tr>
<td><code>-bd</code></td>
<td>Runs in daemon mode and listens for connections on port 25</td>
</tr>
<tr>
<td><code>-bD</code></td>
<td>Runs in daemon mode, but in the foreground rather than the background(^a)</td>
</tr>
<tr>
<td><code>-bh</code></td>
<td>Views recent connection info (same as <code>hoststat</code>)</td>
</tr>
<tr>
<td><code>-bH</code></td>
<td>Purges disk copy of outdated connection info (same as <code>purgestat</code>)</td>
</tr>
<tr>
<td><code>-bi</code></td>
<td>Initializes hashed aliases (same as <code>newaliases</code>)</td>
</tr>
<tr>
<td><code>-bm</code></td>
<td>Runs as a mailer, delivers mail in the usual way (default)</td>
</tr>
<tr>
<td><code>-bp</code></td>
<td>Prints the mail queue (same as <code>mailq</code>)</td>
</tr>
<tr>
<td><code>-bP</code></td>
<td>Prints the number of entries in queues via shared memory</td>
</tr>
<tr>
<td><code>-bs</code></td>
<td>Enters SMTP server mode (on standard input, not port 25)</td>
</tr>
<tr>
<td><code>-bt</code></td>
<td>Enters address test mode</td>
</tr>
<tr>
<td><code>-bv</code></td>
<td>Verifies mail addresses only; doesn’t send mail</td>
</tr>
</tbody>
</table>

\(^a\) Use this mode for debugging so you can see the error and debugging messages.

If you are configuring a server that will accept incoming mail from the Internet, run `sendmail` in daemon mode (`-bd`). In this mode, `sendmail` listens on network port 25 and waits for work. (The ports that `sendmail` listens on are determined by `DAEMON_OPTIONS`; port 25 is the default.)

You will usually specify the `-q` flag, too—it sets the interval at which `sendmail` processes the mail queue. For example, `-q30m` runs the queue every thirty minutes and `-q1h` runs it every hour.

`sendmail` normally tries to deliver messages immediately, saving them in the queue only momentarily to guarantee reliability. But if your host is too busy or the destination machine is unreachable, `sendmail` queues messages and tries to send them again later. `sendmail` uses persistent queue runners that are usually started at boot time. It does locking, so multiple, simultaneous queue runs are safe. You can use the “queue groups” configuration feature to facilitate delivery of large mailing lists and queues.
sendmail reads its configuration file, sendmail.cf, only when it starts up. Therefore, you must either kill and restart sendmail or send it a HUP signal when you change the config file. sendmail creates a sendmail.pid file that contains its process ID and the command that started it. You should start sendmail with an absolute path because it re-exec's itself on receipt of the HUP signal. The sendmail.pid file allows the process to be HUPed with the command

```bash
sudo kill -HUP 'head -1 sendmail.pid'
```

The location of the PID file is OS dependent. It’s usually /var/run/sendmail.pid or /etc/mail/sendmail.pid but can be set in the config file with the confPID_FILE option:

```plaintext
define(confPID_FILE, '/var/run/sendmail.pid')
```
Mail queues

Sendmail uses at least two queues: /var/spool/mqueue when acting as an MTA on port 25, and /var/spool/clientmqueue when acting as an MSA on port 587. Sendmail can use multiple queues beneath mqueue to increase performance. All messages make at least a brief stop in the queue before being sent on their way.

A queued message is saved in pieces in several different files. Table 18.6 shows the six possible pieces. Each filename has a two-letter prefix that identifies the piece, followed by a random ID built from sendmail’s process ID.

Table 18.6: Prefixes for files in the mail queue

<table>
<thead>
<tr>
<th>Prefix</th>
<th>File contents</th>
</tr>
</thead>
<tbody>
<tr>
<td>qf</td>
<td>The message header and control file</td>
</tr>
<tr>
<td>df</td>
<td>The body of the message</td>
</tr>
<tr>
<td>tf</td>
<td>A temporary version of the qf file while the qf file is being updated</td>
</tr>
<tr>
<td>Tf</td>
<td>A notice that 32 or more failed locking attempts have occurred</td>
</tr>
<tr>
<td>Qf</td>
<td>A notice that the message bounced and could not be returned</td>
</tr>
<tr>
<td>xf</td>
<td>Temporary transcript file of error messages from mailers</td>
</tr>
</tbody>
</table>

If subdirectories qf, df, or xf exist in a queue directory, then those pieces of the message are put in the proper subdirectory. The qf file contains not only the message header but also the envelope addresses, the date at which the message should be returned as undeliverable, the message’s priority in the queue, and the reason the message is in the queue. Each line begins with a single-letter code that identifies the rest of the line.

Each message that is queued must have a qf and df file. All the other prefixes are used by sendmail during attempted delivery. When a machine crashes and reboots, the startup sequence for sendmail should delete the tf, xf, and Tf files from each queue. If you are the sysadmin responsible for mail, check occasionally for Qf files in case local configuration is causing the bounces. An occasional glance at the queue directories lets you spot problems before they become disasters.

The mail queue opens up several opportunities for things to go wrong. For example, the filesystem can fill up (avoid putting /var/spool/mqueue and /var/log on the same partition), the queue can become clogged, or orphaned mail messages can get stuck in the queue. Sendmail has configuration options to help with performance on busy machines.
**sendmail configuration**

sendmail is controlled by a single configuration file, typically called `/etc/mail/sendmail.cf` for a sendmail running as an MTA or `/etc/mail/submit.cf` for a sendmail acting as an MSA. The flags with which sendmail is started determine which config file it uses: `-bm`, `-bs`, and `-bt` use `submit.cf` if it exists, and all other modes use `sendmail.cf`. You can change these names with command-line flags or config file options, but it is best not to.

The raw config file format was designed to be easy to parse by machines, not humans. The `m4` source (`.`mc) file from which the `.cf` file is generated is an improvement, but its picky and rigid syntax isn’t going to win any awards for user friendliness either. Fortunately, many of the paradigms you might want to set up have already been hammered out by others with similar needs and are supplied in the distribution as prepackaged features.

**sendmail** configuration involves several steps:

1. Determine the role of the machine you are configuring: client, server, Internet-facing mail receiver, etc.

2. Choose the features needed to implement that role and build an `.mc` file for the configuration

3. Compile the `.mc` file with `m4` to produce a `.cf` config file

We cover the features commonly used for site-wide, Internet-facing servers and for little desktop clients. For more detailed coverage, we refer you to two key pieces of documentation on the care and feeding of **sendmail**: the O’Reilly book *sendmail* by Bryan Costales et al. and the file `cf/README` from the distribution.
The m4 preprocessor

m4, originally intended as a front end for programming languages, lets users write more readable (or perhaps more cryptic) programs. m4 is powerful enough to be useful in many input transformation situations, and it works nicely for sendmail configuration files.

m4 macros have the form

name(arg1, arg2, ..., argn)

There cannot be any space between the name and the opening parenthesis. Left and right single quotes (that is, backticks and “normal” single quotes) designate strings as arguments. m4’s quote conventions are weird, since the left and right quotes are different characters. Quotes nest, too.

m4 has some built-in macros, and users can also define their own. Table 18.7 lists the most common built-in macros that are used in sendmail configuration.

Table 18.7: m4 macros commonly used with sendmail

<table>
<thead>
<tr>
<th>Macro</th>
<th>Function</th>
</tr>
</thead>
<tbody>
<tr>
<td>define</td>
<td>Defines a macro named arg1 with value arg2</td>
</tr>
<tr>
<td>divert</td>
<td>Manages output streams</td>
</tr>
<tr>
<td>dn1</td>
<td>Discards characters up to and including the next newline</td>
</tr>
<tr>
<td>include</td>
<td>Includes (interpolates) the file named arg1</td>
</tr>
<tr>
<td>undefined</td>
<td>Discards a previous definition of macro named arg1</td>
</tr>
</tbody>
</table>
The sendmail configuration pieces

The *sendmail* distribution includes a *cf* subdirectory beneath which are all the pieces necessary for *m4* configuration. Table 18.4 shows the location of the *cf* directory if you did not install the *sendmail* source but relied on your vendor. The *README* file found in the *cf* directory is *sendmail*’s configuration documentation. The subdirectories, listed in Table 18.8, contain examples and snippets you can include in your own configuration.

<table>
<thead>
<tr>
<th>Directory</th>
<th>Contents</th>
</tr>
</thead>
<tbody>
<tr>
<td>cf</td>
<td>Sample <em>.mc</em> (master configuration) files</td>
</tr>
<tr>
<td>domain</td>
<td>Sample <em>m4</em> files for various domains at Berkeley</td>
</tr>
<tr>
<td>feature</td>
<td>Fragments that implement various features</td>
</tr>
<tr>
<td>hack</td>
<td>Special features of dubious value or implementation</td>
</tr>
<tr>
<td>m4</td>
<td>The basic config file and other core files</td>
</tr>
<tr>
<td>mailer</td>
<td><em>m4</em> files that describe common mailers (delivery agents)</td>
</tr>
<tr>
<td>ostype</td>
<td>OS-dependent file locations and quirks</td>
</tr>
<tr>
<td>sh</td>
<td>Shell scripts used by <em>m4</em></td>
</tr>
</tbody>
</table>

The *cf/cf* directory contains examples of *.mc* files. In fact, it contains so many examples that yours may get lost in the clutter. We recommend that you keep your own *.mc* files separate from those in the distributed *cf* directory. Either create a new directory named for your site (*cf/sitename*) or move the *cf* directory aside to *cf/examples* and create a new *cf* directory. If you do this, copy the *Makefile* and *Build* script over to your new directory so the instructions in the *README* file still work. Alternatively, you can copy all your own configuration *.mc* files to a central location rather than leaving them inside the *sendmail* distribution. The *Build* script uses relative pathnames, so you’ll have to modify it if you want to build a *cf* file from an *.mc* file and are not in the *sendmail* distribution hierarchy.

The files in the *cf/ostype* directory configure *sendmail* for each specific operating system. Many are predefined, but if you have moved things around on your system, you might have to modify one or create a new one. Copy one that is close to reality for your system and give it a new name.

The *cf/feature* directory is where you shop for any configuration pieces you might need. There is a feature for just about anything that any site running *sendmail* has found useful.

The other directories beneath *cf* are pretty much boilerplate and do not need to be tweaked or even understood—just use them.
A configuration file built from a sample .mc file

Before we take off into the wilds of the various configuration macros, features, and options you might use in a sendmail configuration, we shall put the cart before the horse and devise a “no frills” configuration to illustrate the general process. Our example is for a leaf node, myhost.example.com; the master configuration file is called myhost.mc. Here’s the complete .mc file:

```plaintext
divid(-1)
### basic .mc file for example.com
divid(0)
VERSIONID('Id$')
OSTYPE('linux')
MAILER('local')
MAILER('smtp')
```

Except for the diversions and comments, each line invokes a prepackaged macro. The first four lines are boilerplate; they insert comments in the compiled file to note the version of sendmail, the directory the configuration was built in, etc. The OSTYPE macro includes the ..ostype/linux.m4 file. The MAILER lines allow for local delivery (to users with accounts on myhost.example.com) and for delivery to Internet sites.

To build the real configuration file, just run the Build command you copied over to the new cf directory:

```
$ ./Build myhost.cf
```

Finally, install myhost.cf in the right spot—normally /etc/mail/sendmail.cf, but some vendors move it. Favorite vendor hiding places are /etc and /usr/lib.

At a larger site, you might want to create a separate m4 file to hold site-wide defaults; put it in the cf/domain directory. Individual hosts can then include the contents of this file with the DOMAIN macro. Not every host needs a separate config file, but each group of similar hosts (same architecture and same role: server, client, etc.) will probably need its own configuration.

The order of the macros in the .mc file is not arbitrary. It should be

```plaintext
VERSIONID
OSTYPE
DOMAIN
FEATURE
local macro definitions
MAILER
```

Even with sendmail’s easy m4 configuration system, you still have to make several configuration decisions for your site. As you read about the features described below, think about
how they might fit into your site’s organization. A small site will probably have only a hub node and leaf nodes and thus will need only two versions of the config file. A larger site might need separate hubs for incoming and outgoing mail and, perhaps, a separate POP/IMAP server.

Whatever the complexity of your site and whatever face it shows to the outside world (exposed, behind a firewall, or on a virtual private network, for example), it’s likely that the cf directory contains some appropriate ready-made configuration snippets just waiting to be customized and put to work.
Configuration primitives

sendmail configuration commands are case sensitive. By convention, the names of predefined macros are all caps (e.g., OSTYPE), m4 commands are all lower case (e.g., define), and configurable option names usually start with lowercase conf and end with an all-caps variable name (e.g., confFAST_SPLIT). Macros usually refer to an m4 file called ../macroname/arg1.m4. For example, the reference OSTYPE('linux') causes the file ../ostype/linux.m4 to be included.
Tables and databases

Before we plunge into specific configuration primitives, we must first discuss tables (sometimes called maps or databases), which sendmail can use to perform mail routing or address rewriting. Most are used in conjunction with the feature macro.

A table is a cache (usually a text file) of routing, aliasing, policy, or other information that is converted to a database format with the makemap command and then used as an information source for one or more of sendmail’s various lookup operations. Although the data usually starts as a text file, data for sendmail tables can come from DNS, LDAP, or other sources. The use of a centralized IMAP server relieves sendmail of the chore of chasing down users and obsoletes some of its tables.

sendmail defines three database map types:

- dbm – legacy; uses an extensible hashing algorithm (dbm/ndbm)
- hash – uses a standard hashing scheme (DB)
- btree – uses a B-tree data structure (DB)

For most table applications in sendmail, the hash database type—the default—is the best. Use the makemap command to build the database file from a text file; you specify the database type and the output file base name. The text version of the database should appear on makemap’s standard input. For example:

$ sudo makemap hash /etc/mail/access < /etc/mail/access

At first glance this command looks like a mistake that would cause the input file to be overwritten by an empty output file. However, makemap tacks on an appropriate suffix, so the actual output file is /etc/mail/access.db and in fact no conflict occurs. Each time the text file is changed, the database file must be rebuilt with makemap (but sendmail need not be HUP’d).

Comments can appear in the text files from which maps are produced. They begin with # and continue until the end of the line.

In most circumstances, the longest possible match is used for database keys. As with any hashed data structure, the order of entries in the input text file is not significant. Some features expect a database file as a parameter; they default to hash as the database type and /etc/mail/tablename.db as the filename for the database.
Generic macros and features

Table 18.9 lists common configuration primitives, whether they are typically used (yes, no, maybe), and a brief description of what they do.

Table 18.9: sendmail generic configuration primitives

<table>
<thead>
<tr>
<th>Primitive</th>
<th>Used?</th>
<th>Description</th>
</tr>
</thead>
<tbody>
<tr>
<td>OSTYPE</td>
<td>Yes</td>
<td>Includes OS-specific paths and mailer flags</td>
</tr>
<tr>
<td>DOMAIN</td>
<td>No</td>
<td>Includes site-specific configuration details</td>
</tr>
<tr>
<td>MAILER</td>
<td>Yes</td>
<td>Enables mailers, typically smtp and local</td>
</tr>
<tr>
<td>FEATURE</td>
<td>Maybe</td>
<td>Enables a variety of sendmail features</td>
</tr>
<tr>
<td>use_cw_file</td>
<td>Yes (S)</td>
<td>Lists hosts for which you accept mail</td>
</tr>
<tr>
<td>redirect</td>
<td>Maybe (S)</td>
<td>Bounces mail nicely when users move</td>
</tr>
<tr>
<td>always_add_domain</td>
<td>Yes</td>
<td>Fully qualifies hostnames if UA didn’t</td>
</tr>
<tr>
<td>access_db</td>
<td>Maybe (S)</td>
<td>Sets database of hosts to relay mail for</td>
</tr>
<tr>
<td>virtusertable</td>
<td>Maybe (S)</td>
<td>Turns on domain aliasing (virtual domains)</td>
</tr>
<tr>
<td>ldap_routing</td>
<td>Maybe (S)</td>
<td>Routes incoming mail using LDAP</td>
</tr>
<tr>
<td>MASQUERADE_AS</td>
<td>Yes</td>
<td>Makes all mail seem to come from one place</td>
</tr>
<tr>
<td>EXPOSED_USER</td>
<td>Yes</td>
<td>Lists users who shouldn’t be masqueraded</td>
</tr>
<tr>
<td>MAIL_HUB</td>
<td>Yes (S)</td>
<td>Specifies mail server for incoming mail</td>
</tr>
<tr>
<td>SMART_HOST</td>
<td>Yes (C)</td>
<td>Specifies mail server for outgoing mail</td>
</tr>
</tbody>
</table>

a. S = servers, C = clients

OSTYPE macro

An OSTYPE file packages a variety of vendor-specific information, such as the expected locations of mail-related files, paths to commands that sendmail needs, flags to mailer programs, etc. See cf/README for a list of all the variables that can be defined in an OSTYPE file.

So where is the OSTYPE macro itself defined? In a file in the cf/m4 directory, which is magically prepended to your config file when you run the Build script.

DOMAIN macro

The DOMAIN directive lets you specify site-wide generic information in one place (cf/domain/filename.m4) and then include it in each host’s config file with

DOMAIN(‘filename’)

MAILER macro
You must include a `MAILER` macro for every delivery agent you want to enable. You’ll find a complete list of supported mailers in the directory `cf/mailers`, but typically you need only `local` and `smtp`. `MAILER` lines are generally the last thing in the `.mc` file.

**FEATURE macro**

The `FEATURE` macro enables a whole host of common scenarios (56 at last count!) by including `m4` files from the `feature` directory. The syntax is

```
FEATURE(keyword, arg, arg, ...)
```

where `keyword` corresponds to a file `keyword.m4` in the `cf/feature` directory and the `args` are passed to it. There can be at most nine arguments to a feature.

**use_cw_file feature**

The `sendmail` internal class `w` (hence the name `cw`) contains the names of all local hosts for which this host accepts and delivers mail. This feature specifies that mail be accepted for the hosts listed, one per line, in `/etc/mail/local-host-names`. The configuration line

```
FEATURE('use_cw_file')
```

invokes the feature. A client machine does not really need this feature unless it has nicknames, but your incoming mail hub machine does. The `local-host-names` file should include any local hosts and virtual domains for which you accept email, including sites whose backup MX records (see this page) point to you.

Without this feature, `sendmail` delivers mail locally only if it is addressed to the machine on which `sendmail` is running.

If you add a new host at your site, you must add it to the `local-host-names` file and send a HUP signal to `sendmail` to make your changes take effect.

**redirect feature**

When people leave your organization, you usually either forward their mail or let mail to them bounce back to the sender with an error. The `redirect` feature provides support for a more elegant way of bouncing mail.

If Joe Smith has graduated from oldsite.edu (login smithj) to newsite.com (login joe), then enabling `redirect` with

```
FEATURE('redirect')
```

and adding the line

```
smithj: joe@newsite.com.REDIRECT
```
to the **aliases** file at oldsite.edu causes mail to smithj to be returned to the sender with an error message suggesting that the sender try the address joe@newsite.com instead. The message itself is not automatically forwarded.

*always_add_domain feature*

The *always_add_domain* feature makes all email addresses fully qualified. It should always be used.

*access_db feature*

The *access_db* feature controls relaying and other policy issues. Typically, the raw data that drives this feature either comes from LDAP or is kept in a text file called `/etc/mail/access`. In the latter case, the text file must be converted to some kind of indexed format with the `makemap` command, as described [here](#). To use the flat file, use `FEATURE('access_db')` in the configuration file; for the LDAP version, use `FEATURE('access_db', 'LDAP')`. The LDAP form uses the default schema defined in the file `cf/sendmail.schema`; if you want a different schema file, use additional arguments in your `FEATURE` statement.

The key field in the access database is an IP network or a domain name with an optional tag such as `Connect:`, `To:`, or `From:`. The value field specifies what to do with the message.

The most common values are **OK** to accept the message, **RELAY** to allow it to be relayed, **REJECT** to reject it with a generic error indication, or **ERROR:**"*error code and message*" to reject it with a specific message. Other possible values allow for finer-grained control. Here is a snippet from a sample `/etc/mail/access` file:

```
localhost       RELAY
127.0.0.1        RELAY
192.168.1.1      RELAY
192.168.1.17     RELAY
66.77.123.1      OK
fax.com          OK
61               ERROR:"550 We don't accept mail from spammers"
67.106.63        ERROR:"550 We don't accept mail from spammers"
```

*virtusertable feature*

The **virtusertable** feature supports domain aliasing for incoming mail through a map stored in `/etc/mail/virtusertable`. This feature lets one machine host multiple virtual domains and is used frequently at web-hosting sites. The key field of the table contains either an email address (`user@host.domain`) or a domain specification (`@domain`). The value field is a local or external email address. If the key is a domain, the value can either pass the `user` field along as the variable `%1` or route the mail to a different user. Here are some examples:
All the host keys on the left side of the data mappings must be listed in the cw file, 
/etc/mail/local-host-names, or be included in the VIRTUSER_DOMAIN list. If they are not, sendmail will not know to accept the mail locally and will try to find the destination host on the Internet. But DNS MX records will point sendmail back to this same server and you will get a “local configuration error” message in the resulting bounce message. Unfortunately, sendmail cannot tell that the error message for this instance should in fact be “virtusertable key not in cw file.”

ldap_routing feature

LDAP, the Lightweight Directory Access Protocol, can be a source of data for aliases or mail routing information as well as general tabular data as described earlier. The cf/README file has a long section on LDAP with lots of examples.

See this page for general information about LDAP.

To use LDAP in this way, you must have built sendmail to include LDAP support. In your .mc file, add the lines

```bash
define('confLDAP_DEFAULT_SPEC', '-h server -b searchbase')
FEATURE('ldap_routing')
LDAPROUTE_DOMAIN('my_domain')
```

Those lines tell sendmail that you want to use an LDAP database to route incoming mail addressed to the specified domain. The LDAP_DEFAULT_SPEC option identifies the LDAP server and the LDAP basename for searches. LDAP uses port 389 unless you specify a different port by adding -p ldap_port to the define.

sendmail uses the values of two tags in the LDAP database:

- mailLocalAddress for the addressee on incoming mail
- mailRoutingAddress for the destination to which email should be sent

sendmail also supports the tag mailHost, which if present routes mail to the MX-designated mail handler for the specified host. The recipient address remains the value of the mailRoutingAddress tag.

LDAP database entries support a wild card entry, @domain, that reroutes mail addressed to anyone at the specified domain (as was done in the virtusertable).
By default, mail addressed to user@host1.mydomain would first trigger a lookup on user@host1.mydomain. If that failed, sendmail would try @host1.mydomain but not user@mydomain. Including the line

LDAPROUTE_EQUIVALENT('host1.mydomain')

would also try the keys user@mydomain and @mydomain. This feature enables a single database to route mail at a complex site. You can also take the entries for the LDAPROUTE_EQUIVALENT clauses from a file, which makes the feature quite usable. The syntax for that form is

LDAPROUTE_EQUIVALENT_FILE('filename')

Additional arguments to the ldap_routing feature let you specify more details about the LDAP schema and control the handling of addressee names that have a +detail part. As always, see the cf/README file for exact details.

Masquerading features

An email address is usually made up of a username, a host, and a domain, but many sites do not want the names of their internal hosts exposed on the Internet. The MASQUERADE_AS macro lets you specify a single identity for other machines to hide behind. All mail appears to emanate from the designated machine or domain. This is fine for regular users, but for debugging purposes, system users such as root should be excluded from the masquerade.

For example, the sequence

MASQUERADE_AS('atrust.com')
EXPOSED_USER('root')
EXPOSED_USER('Mailer-Daemon')

would stamp mail as coming from user@atrust.com unless it was sent by root or the mail system; in these cases, the mail would carry the name of the originating host.

MASQUERADE_AS is actually just the tip of a vast masquerading iceberg that extends downward through a dozen variations and exceptions. The allmasquerade and masquerade_envelope features (in combination with MASQUERADE_AS) hide just the right amount of local info. See the cf/README for details.

MAIL_HUB and SMART_HOST macros

Masquerading makes all mail appear to come from a single host or domain by rewriting the headers and, optionally, the envelope. But most sites want all mail to actually come from (or go to) a single machine so that they can control the flow of viruses, spam, and company secrets. You can achieve this control with a combination of MX records in DNS, the MAIL_HUB macro for incoming mail, and the SMART_HOST macro for outgoing mail.
For example, in a structured email implementation, MX records would direct incoming email from the Internet to an MTA in the network’s demilitarized zone. After verification that the received email was free of viruses and spam and was directed to valid local users, the mail could be relayed, with the following `define`, to the internal routing MTA for delivery:

```plaintext
define('MAIL_HUB', 'smtp:routingMTA.mydomain')
```

Likewise, client machines would relay their mail to the `SMART_HOST` designated in the `nullclient` feature in their configuration. The `SMART_HOST` could then filter for viruses and spam so that mail from your site did not pollute the Internet.

The syntax of `SMART_HOST` parallels that of `MAIL_HUB`, and the default delivery agent is again `relay`. For example:

```plaintext
define('SMART_HOST', 'smtp:outgoingMTA.mydomain')
```

You can use the same machine as the server for both incoming and outgoing mail. Both the `SMART_HOST` and the `MAIL_HUB` must allow relaying, the first from clients inside your domain and the second from the MTA in the DMZ.
Client configuration

Most of your site’s machines should be configured as clients who just submit outgoing mail generated by users and don’t receive mail at all. One of sendmail’s FEATURES, nullclient, is just right for this situation. It creates a config file that forwards all mail to a central hub over SMTP. The entire config file, after the VERSIONID and OSTYPE lines, would be simply

```
FEATURE('nocanonify')
FEATURE('nullclient', 'mailserver')
EXPOSED_USER('root')
```

where mailserver is the name of your central hub. The nocanonify feature tells sendmail not to do DNS lookups or rewrite addresses with fully qualified domain names. All that work will be done by the mailserver host. This feature is similar to SMART_HOST and assumes that the client will MASQUERADE_AS mailserver. The EXPOSED_USER clause exempts root from the masquerading and so facilitates debugging.

The mailserver machine must allow relaying from its null clients. That permission is granted in the access_db, described here. The null client must have an associated MX record that points to mailserver and must also be included in the mailserver’s cw file (usually /etc/mail/local-host-names). These settings allow the mailserver to accept mail for the client.

sendmail should run as an MSA (without the -bd flag) if the user agents on the client machine can be taught to use port 587 for submitting mail. If not, you can run sendmail in daemon mode (-bd), but set the DAEMON_OPTIONS configuration option to listen for connections only on the loopback interface.
m4 configuration options

You set config file options with the m4 define command. A complete list of options that are accessible as m4 variables (along with their default values) is given in the cf/README file.

The defaults are OK for a typical site that is not too paranoid about security and not too concerned with performance. The defaults try to protect you from spam by turning off relaying, by requiring addresses to be fully qualified, and by requiring that senders’ domains resolve to an IP address. If your mail hub machine is busy and services a lot of mailing lists, you might need to tweak some of the performance values.

Table 18.10 lists some options that you might need to adjust (about 10% of over 175 configuration options). Their default values are shown in parentheses. To save space, the option names are shown without their conf prefix; for example, the FAST_SPLIT option is actually named confFAST_SPLIT. We divided the table into subsections that identify the kind of issue the variable addresses: resource management, performance, security and spam abatement, and miscellaneous options. Some options fit in more than one category, but we have listed them only once.

Table 18.10: Basic sendmail configuration options
<table>
<thead>
<tr>
<th>Option name</th>
<th>Description (default value)</th>
</tr>
</thead>
<tbody>
<tr>
<td>MAX_DAEMON_CHILDREN</td>
<td>Max number of child processes (no limit)</td>
</tr>
<tr>
<td>MAX_MESSAGE_SIZE</td>
<td>Max size in bytes of a single message (infinite)</td>
</tr>
<tr>
<td>MIN_FREE_BLOCKS</td>
<td>Min filesystem space to accept mail (100)</td>
</tr>
<tr>
<td>TO_lots_of_stuff</td>
<td>Timeouts for all kinds of things (various)</td>
</tr>
<tr>
<td>DELAY_LA</td>
<td>Load avg. to slow deliveries (0 = no limit)</td>
</tr>
<tr>
<td>FAST_SPLIT</td>
<td>Suppresses MX lookups as recipients are sorted and split across queues (1 = true)</td>
</tr>
<tr>
<td>MCI_CACHE_SIZE</td>
<td># of open outgoing TCP connections cached (2)</td>
</tr>
<tr>
<td>MCI_CACHE_TIMEOUT</td>
<td>Time to keep cached connections open (5m)</td>
</tr>
<tr>
<td>MIN_QUEUE_TIMEOUT</td>
<td>Minimum time jobs must stay in queue (0)</td>
</tr>
<tr>
<td>QUEUE_LA</td>
<td>Load average at which mail should be queued instead of delivered immediately (8 * #CPUs)</td>
</tr>
<tr>
<td>REFUSE_LA</td>
<td>Load avg. at which to refuse mail (12 * #CPUs)</td>
</tr>
<tr>
<td>AUTH_MECHANISMS</td>
<td>SMTP auth mechanisms (Table 18.10)</td>
</tr>
<tr>
<td>CONNECTION_RATE_THROTTLE</td>
<td>Limits connection acceptance rate (no limit)</td>
</tr>
<tr>
<td>DONT_BLAKE_SENDMAIL</td>
<td>Overrides security and file checking (safe)</td>
</tr>
<tr>
<td>MAX_MIME_HEADER_LENGTH</td>
<td>Sets max size of MIME headers (no limit)</td>
</tr>
<tr>
<td>MAX_RCPTS_PER_MESSAGE</td>
<td>Slows spam delivery; defers extra recipients and sends a temporary error msg (infinite)</td>
</tr>
<tr>
<td>PRIVACY_FLAGS</td>
<td>Limits info given out by SMTP (authwarnings)</td>
</tr>
<tr>
<td>DOUBLE_BOUNCE_ADDRESS</td>
<td>Catches lots of spam; some sites use /dev/null, which can hide serious problems (postmaster)</td>
</tr>
<tr>
<td>LDAP_DEFAULT_SPEC</td>
<td>Map spec for LDAP database, including the host/port the server is running on (undefined)</td>
</tr>
</tbody>
</table>

---

a. More specifically, the maximum number of child processes that can run at once. When the limit is reached, sendmail refuses connections. This option can prevent (or create) denial of service attacks.

b. The default value is EXTERNAL GSSAPI KERBEROS_V4 DIGEST-MD5 CRAM-MD5; don’t add PLAIN LOGIN, because the password is transmitted as cleartext. That might be OK internally, but not on the Internet unless the connection is also secured through the use of SSL.

c. Don’t change this setting casually!

d. This option can prevent user agent buffer overflows. “256/128” is a good value to use—it means 256 bytes per header and 128 bytes per parameter to that header.
Spam-related features in sendmail

sendmail has a variety of features and configuration options that can help you control spam and viruses:

- Rules that control third party (aka promiscuous, aka open) relaying; that is, the use of your mail server by one off-site user to send mail to another off-site user. Spammers often use relaying to mask the true source of their mail and thereby avoid detection by ISPs. Relaying also lets spammers use your cycles and save their own.

- The access database for filtering recipient addresses. This feature is rather like a firewall for email.

- Blacklists that catalog open relays and known spam-friendly sites that sendmail can check against.

- Throttles that can slow down mail acceptance when certain types of bad behavior are detected.

- Header checking and input mail filtering by means of a generic mail filtering interface called libmilter. It allows arbitrary scanning of message headers and content and lets you reject messages that match a particular profile. Milters are plentiful and powerful; see milter.org.

Relay control

sendmail accepts incoming mail, looks at the envelope addresses, decides where the mail should go, and then passes the message along to an appropriate destination. That destination can be local or it can be another transport agent farther along in the delivery chain. When an incoming message has no local recipients, the transport agent that handles it is said to be acting as a relay.

Only hosts that are tagged with relay in the access database (see this page) or that are listed in /etc/mail/relay-domains are allowed to submit mail for relaying. Some types of relaying are useful and legitimate. How can you tell which messages to relay and which to reject? Relaying is actually necessary in only three situations:

- When the transport agent acts as a gateway for hosts that are not reachable in any other way; for example, hosts that are not always turned on (laptops, Windows PCs) and virtual hosts. In this situation, all the recipients for which you want to relay lie within the same domain.

- When the transport agent is the outgoing mail server for other, not-so-smart hosts. In this case, all the senders’ hostnames or IP addresses are local (or at least enumerable).

- When you have agreed to be a backup MX destination for another site.
Any other situation that appears to require relaying is probably just an indication of bad design (with the possible exception of support for mobile users). You can obviate the first use of relaying (above) by designating a centralized server to receive mail, with POP or IMAP being used for client access. The second case should always be allowed, but only for your own hosts. You can check IP addresses or hostnames. In the third case, you can list the other site in your access database and allow relaying just for that site’s IP address blocks.

Although sendmail comes with relaying turned off by default, several features can turn relaying back on, either fully or in a limited and controlled way. These features are listed below for completeness, but our recommendation is that you be careful about opening things up too much. The access_db feature is the safest way to allow limited relaying.

- FEATURE(‘relay_entire_domain’) – allows relaying for just your domain
- RELAY_DOMAIN(‘domain, ...’) – adds more domains to be relayed
- RELAY_DOMAIN_FILE(‘filename’) – same; takes domain list from a file
- FEATURE(‘relay_hosts_only’) – affects RELAY_DOMAIN, accessdb

You need to make an exception if you use the SMART_HOST or MAIL_HUB designations to route mail through a particular mail server machine. That server must be set up to relay mail from local hosts. Configure it with

FEATURE(‘relay_entire_domain’)

If you consider turning on relaying in some form, consult the sendmail documentation in cf/README to be sure you don’t inadvertently become a friend of spammers. When you are done, have one of the relay-checking sites verify that you did not inadvertently create an open relay—try spamhelp.org.

**User or site blacklisting**

If you have local users or hosts to which you want to block mail, use

FEATURE(‘blacklist_recipients’)

It supports the following types of entries in your access file:

```
To:nobody@          ERROR:550 Mailbox disabled for this user
To:printer.mydomain ERROR:550 This host does not accept mail
To:user@host.mydomain ERROR:550 Mailbox disabled for this user
```

These lines block incoming mail to user nobody on any host, to host printer, and to a particular user’s address on one machine. The use of the To: tag lets these users send messages, just not receive them; some printers have that capability.
To include a DNS-style blacklist for incoming email, use the `dnsbl` feature:

```
FEATURE('dnsbl', 'zen.spamhaus.org')
```

This feature makes `sendmail` reject mail from any site whose IP address is in any of the three blacklists of known spammers (SBL, XBL, and PBL) maintained at spamhaus.org. Other lists catalog sites that run open relays and blocks of addresses that are known to be havens for spammers. These blacklists are distributed through a clever tweak of the DNS system; hence the name `dnsbl`.

You can pass a third argument to the `dnsbl` feature to specify the error message you would like returned. If you omit this argument, `sendmail` returns a fixed error message from the DNS database that contains the records.

You can include the `dnsbl` feature several times to check multiple lists of abusers.

**Throttles, rates, and connection limits**

Table 18.11 lists several `sendmail` controls that can slow down mail processing when clients’ behavior appears suspicious.

<table>
<thead>
<tr>
<th>Primitive</th>
<th>Description</th>
</tr>
</thead>
<tbody>
<tr>
<td>BAD_RCPT_THROTTLE</td>
<td>Slows down spammers collecting addresses</td>
</tr>
<tr>
<td>MAX_RCPTS_PER_MESSAGE</td>
<td>Defers delivery if a message has too many recipients</td>
</tr>
<tr>
<td>ratecontrol feature</td>
<td>Limits the rate of incoming connections</td>
</tr>
<tr>
<td>conncontrol feature</td>
<td>Limits the number of simultaneous connections</td>
</tr>
<tr>
<td>greet_pause feature</td>
<td>Delays HELO response, forces strict SMTP compliance</td>
</tr>
</tbody>
</table>

After the no-such-login count reaches the limit set in the `BAD_RCPT_THROTTLE` option, `sendmail` sleeps for one second after each rejected `RCPT` command, slowing a spammer’s address harvesting to a crawl. To set that threshold to 3, use

```
define('confBAD_RCPT_THROTTLE', '3')
```

Setting the `MAX_RCPTS_PER_MESSAGE` option causes the sender to queue extra recipients for later. This is a cheap form of greylisting for messages that have a suspiciously large number of recipients.

The `ratecontrol` and `conncontrol` features allow per-host or per-net limits on the rate at which incoming connections are accepted and the number of simultaneous connections, respectively. Both use the `/etc/mail/access` file to specify the limits and the domains to which they should apply, the first
with the tag ClientRate: in the key field and the second with tag ClientConn:. To enable rate controls, insert lines like these in your .mc file:

```plaintext
FEATURE('access_db')
FEATURE('ratecontrol', 'nodelay', 'terminate')
FEATURE('conncontrol', 'nodelay', 'terminate')
```

Then, add to your /etc/mail/access file the list of hosts or nets to be controlled and their restriction thresholds. For example, the lines

```
ClientRate:192.168.6.17  2
ClientRate:170.65.3.4    10
```

limit the hosts 192.168.6.17 and 170.65.3.4 to two new connections per minute and ten new connections per minute, respectively. The lines

```
ClientConn:192.168.2.8    2
ClientConn:175.14.4.1     7
ClientConn:              10
```

set limits of two simultaneous connections for 192.168.2.8, seven for 175.14.4.1, and ten simultaneous connections for all other hosts.

Another nifty feature is greet_pause. When a remote transport agent connects to your sendmail server, the SMTP protocol mandates that it wait for your server’s welcome greeting before speaking. However, it’s common for spam mailers to blurt out an EHLO/HELO command immediately. This behavior is partially explainable as poor implementation of the SMTP protocol in spam-sending tools, but it may also be a feature that aims to save time on the spammer’s behalf. Whatever the cause, this behavior is suspicious and is known as “slamming.”

The greet_pause feature makes sendmail wait for a specified period of time at the beginning of the connection before greeting its newfound friend. If the remote MTA does not wait to be properly greeted and proceeds with an EHLO or HELO command during the planned awkward moment, sendmail logs an error and refuses subsequent commands from the remote MTA.

You can enable greeting pauses with this entry in the .mc file:

```plaintext
FEATURE('greet_pause', '700')
```

This line causes a 700 millisecond delay at the beginning of every new connection. You can set per-host or per-net delays with a GreetPause: prefix in the access database, but most sites use a blanket value for this feature.
Security and sendmail

With the explosive growth of the Internet, programs such as sendmail that accept arbitrary user-supplied input and deliver it to local users, files, or shells have frequently provided an avenue of attack for hackers. sendmail, along with DNS and even IP, is flirting with authentication and encryption as a built-in solution to some of these fundamental security issues.

sendmail supports both SMTP authentication and encryption with TLS, Transport Layer Security (formerly known as SSL, the Secure Sockets Layer). TLS brought with it six new configuration options for certificate files and key files. New actions for access database matches can require that authentication must have succeeded.

sendmail carefully inspects file permissions before it believes the contents of, say, a .forward or an aliases file. Although this tightening of security is generally welcome, it’s sometimes necessary to relax the tough policies. To this end, sendmail introduced the DontBlameSendmail option, so named in hopes that the name might suggest to sysadmins that what they are doing is unsafe.

This option has many possible values—55 at last count. The default is safe, the strictest possible. For a complete list of values, see doc/op/op.ps in the sendmail distribution or the O’Reilly sendmail book. Or just leave the option set to safe.

Ownerships

Three user accounts are important in the sendmail universe: the DefaultUser, the RunAsUser, and the TrustedUser.

By default, all of sendmail’s mailers run as the DefaultUser unless the mailer’s flags specify otherwise. If a user mailnull, sendmail, or daemon exists in the passwd file, DefaultUser will be that. Otherwise, it defaults to UID 1 and GID 1. We recommend the use of the mailnull account and a mailnull group. Add it to /etc/passwd with a star as the password, no valid shell, no home directory, and a default group of mailnull. You’ll have to add the mailnull entry to the group file, too. The mailnull account should not own any files. If sendmail is not running as root, the mailers must be setuid.

If RunAsUser is set, sendmail ignores the value of DefaultUser and does everything as RunAsUser. If you are running sendmail setgid, then the submission sendmail just passes messages to the real sendmail through SMTP. The real sendmail does not have its setuid bit set, but it runs as root from the startup files.

The RunAsUser is the UID that sendmail runs under after opening its socket connection to port 25. Ports numbered less than 1,024 can be opened only by the superuser; therefore, sendmail must initially run as root. However, after performing this operation, sendmail can switch to a different UID. Such a switch reduces the risk of damage or access if sendmail is tricked into doing something bad. Don’t use the RunAsUser feature on machines that support user accounts or other
services; it is meant for use only on firewalls or bastion hosts (specially hardened hosts intended to withstand attack when placed in a DMZ or outside a firewall).

By default, `sendmail` does not switch identities and continues to run as root. If you change the `RunAsUser` to something other than root, you must change several other things as well. The `RunAsUser` must own the mail queue, be able to read all maps and include files, be able to run programs, etc. Expect to spend a few hours discovering all the file and directory ownerships that must be changed.

`sendmail`’s `TrustedUser` can own maps and alias files. The `TrustedUser` is allowed to start the daemon or rebuild the `aliases` file. This facility exists mostly to support GUI interfaces to `sendmail` that need to provide limited administrative control to certain users. If you set `TrustedUser`, be sure to guard the account that it points to because this account can easily be exploited to gain root access. The `TrustedUser` is different from the `TRUSTED_USERS` class, which determines who can rewrite the From line of messages. (The `TRUSTED_USERS` feature is typically used to support mailing list software.)

**Permissions**

File and directory permissions are important to `sendmail` security. Use the settings listed in Table 18.12 to be safe.

### Table 18.12: Owner and permissions for `sendmail`-related directories

<table>
<thead>
<tr>
<th>Path</th>
<th>Owner</th>
<th>Mode</th>
<th>What it contains</th>
</tr>
</thead>
<tbody>
<tr>
<td><code>/var/spool/clientmqueue</code></td>
<td>smmsp:smmsp</td>
<td>770</td>
<td>Queue for initial submissions</td>
</tr>
<tr>
<td><code>/var/spool/mqueue</code></td>
<td>RunAsUser</td>
<td>700</td>
<td>Mail queue directory</td>
</tr>
<tr>
<td><code>.</code> /var/spool`</td>
<td>root</td>
<td>755</td>
<td>Path to <code>mqueue</code></td>
</tr>
<tr>
<td><code>/etc/mail/*</code></td>
<td>TrustedUser</td>
<td>644</td>
<td>Maps, the config file, aliases</td>
</tr>
<tr>
<td><code>/etc/mail</code></td>
<td>TrustedUser</td>
<td>755</td>
<td>Parent directory for maps</td>
</tr>
<tr>
<td><code>/etc</code></td>
<td>root</td>
<td>755</td>
<td>Path to <code>mail</code> directory</td>
</tr>
</tbody>
</table>

`sendmail` no longer reads `.forward` files that have link counts greater than 1 if the directory paths that lead to them have lax permissions. This rule bit Evi when one of her `.forward` files, which she usually hard-linked to either `.forward.to.boulder` or `.forward.to.sandiego`, silently failed to forward her mail from a small site at which she did not receive much mail. It was months before she realized that “I never got your mail” was her own fault and not a valid excuse.

You can turn off many of the restrictive file access policies mentioned above with the `DontBlameSendmail` option. But don’t do that.

**Safer mail to files and programs**
We recommend that you use **smrsh** instead of **/bin/sh** as your program mailer and that you use **mail.local** instead of **/bin/mail** as your local mailer. Both programs are included in the **sendmail** distribution. To incorporate them into your configuration, add the lines

```plaintext
FEATURE('smrsh', 'path-to-smrsh')
FEATURE('local_lmtmp', 'path-to-mail.local')
```

to your `.mc` file. If you omit the explicit paths, the commands are assumed to live in **/usr/libexec**. You can use **sendmail**'s `confEBINDIR` option to change the default location of the binaries to whatever you want. **Table 18.13** helps you find where our friendly vendors have stashed things.

### Table 18.13: Location of sendmail’s restricted delivery agents

<table>
<thead>
<tr>
<th>OS</th>
<th>smrsh</th>
<th>mail.local</th>
<th>sm.bin</th>
</tr>
</thead>
<tbody>
<tr>
<td>Ubuntu</td>
<td>/usr/lib/sm.bin</td>
<td>/usr/lib/sm.bin</td>
<td>/usr/adm</td>
</tr>
<tr>
<td>Debian</td>
<td>/usr/lib/sm.bin</td>
<td>/usr/lib/sm.bin</td>
<td>/usr/adm</td>
</tr>
<tr>
<td>Red Hat</td>
<td>/usr/sbin</td>
<td>–</td>
<td>/etc/smrsh</td>
</tr>
<tr>
<td>CentOS</td>
<td>/usr/sbin</td>
<td>–</td>
<td>/etc/smrsh</td>
</tr>
<tr>
<td>FreeBSD</td>
<td>/usr/libexec</td>
<td>/usr/libexec</td>
<td>/usr/adm</td>
</tr>
</tbody>
</table>

**smrsh** is a restricted shell that executes only the programs contained in one directory (**/usr/adm/sm.bin** by default). **smrsh** ignores user-specified paths and tries to find any requested commands in its own known-safe directory. **smrsh** also blocks the use of certain shell metacharacters such as `<`, the input redirection symbol. Symbolic links are allowed in **sm.bin**, so you need not make duplicate copies of the programs you allow. The **vacation** program is a good candidate for **sm.bin**. Don’t put **procmail** there; it’s insecure.

Here are some example shell commands and their possible **smrsh** interpretations:

- `vacation eric` # Executes `/usr/adm/sm.bin/vacation eric`
- `cat /etc/passwd` # Rejected, `cat` not in `sm.bin`
- `vacation eric < /etc/passwd` # Rejected, no `<` allowed

**sendmail**’s `SafeFileEnvironment` option controls where files can be written when email is redirected to a file by **aliases** or a **.forward** file. It causes **sendmail** to execute a `chroot` system call, making the root of the filesystem no longer `/` but rather `/safe` or whatever path you specified in the `SafeFileEnvironment` option. An alias that directed mail to the `/etc/passwd` file, for example, would actually be written to `/safe/etc/passwd`.

The `SafeFileEnvironment` option also protects device files, directories, and other special files by allowing writes only to regular files. Besides increasing security, this option ameliorates the effects of user mistakes. Some sites set the option to `/home` to allow access to home directories while keeping system files off-limits.
Mailers can also be run in a **chrooted directory**.

**Privacy options**

*sendmail* privacy options also control

- What external folks can determine about your site through SMTP
- What you require of the host on the other end of an SMTP connection
- Whether your users can see or run the mail queue

Table 18.14 lists the possible values for the privacy options as of this writing; see the file *doc/op/op.ps* in the distribution for current information.

**Table 18.14: Values of the PrivacyOption variable**

<table>
<thead>
<tr>
<th>Value</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>authwarnings</td>
<td>Adds warning header if an outgoing message seems forged</td>
</tr>
<tr>
<td>goaway</td>
<td>Disables all SMTP status queries (EXPN, VRFY, etc.)</td>
</tr>
<tr>
<td>needexpnhelo</td>
<td>Does not expand addresses (EXPN) without a HELO</td>
</tr>
<tr>
<td>needmailhelo</td>
<td>Requires SMTP HELO (identifies remote host)</td>
</tr>
<tr>
<td>needvrfyhelo</td>
<td>Does not verify addresses (VRFY) without a HELO</td>
</tr>
<tr>
<td>nobodyreturn</td>
<td>Does not return the message body in a DSN</td>
</tr>
<tr>
<td>noetra</td>
<td>Disallows asynchronous queue runs</td>
</tr>
<tr>
<td>noexpn</td>
<td>Disallows the SMTP EXPN command</td>
</tr>
<tr>
<td>noreceipts</td>
<td>Turns off delivery status notification for success return receipts</td>
</tr>
<tr>
<td>noverb</td>
<td>Disallows verbose mode for EXPN</td>
</tr>
<tr>
<td>novrfy</td>
<td>Disallows the SMTP VRFY command</td>
</tr>
<tr>
<td>public</td>
<td>Does no privacy/security checking</td>
</tr>
<tr>
<td>restrictexpand</td>
<td>Restricts info displayed by the -bv and -v flags</td>
</tr>
<tr>
<td>restrictmailq</td>
<td>Allows only <em>mqueue</em> directory’s group to see the queue</td>
</tr>
<tr>
<td>restrictqrun</td>
<td>Allows only <em>mqueue</em> directory’s owner to run the queue</td>
</tr>
</tbody>
</table>

---

We recommend conservatism; in your *mc* file, use

```bash
define('confPRIVACY_OPTIONS', '"goaway, authwarnings, restrictmailq, restrictqrun"')```

---

a. ETRN is an ESMTP command for use by dial-up hosts. It requests that the queue be run just for messages to that host.
b. Verbose mode follows .forward files when an EXPN command is given and reports more information on the whereabouts of a user’s mail. Use noverb or, better yet, noexpn, on any machine exposed to the outside world.
c. Unless executed by root or the TrustedUser
sendmail’s default value for the privacy options is authwarnings; the above line would reset that value. Notice the double sets of quotes; some versions of m4 require them to protect the commas in the list of privacy option values.

**Running a chrooted sendmail (for the truly paranoid)**

If you are worried about the access that sendmail has to your filesystem, you can start it in a chrooted jail. Create a minimal filesystem in your jail, including things like /dev/null, /etc essentials (passwd, group, resolv.conf, sendmail.cf, any map files, mail/*), the shared libraries that sendmail needs, the sendmail binary, the mail queue directory, and any log files. You will probably have to fiddle with the list to get it just right. Use the chroot command to start a jailed sendmail. For example:

```
$ sudo chroot /jail /usr/sbin/sendmail -bd -q30m
```

**Denial of service attacks**

Denial of service attacks are difficult to prevent because no a priori method can determine that a message is an attack rather than a valid piece of email. Attackers can try various nasty things, including flooding the SMTP port with bogus connections, filling disk partitions with giant messages, clogging outgoing connections, and mail bombing. sendmail has some configuration parameters that can help slow down or limit the impact of a denial of service attack, but these parameters can also interfere with the delivery of legitimate mail.

The MaxDaemonChildren option limits the number of sendmail processes. It prevents the system from being overwhelmed with sendmail work. However, it also allows an attacker to easily shut down SMTP service.

The MaxMessageSize option can help prevent the mail queue directory from filling. But if you set it too low, legitimate mail will bounce. You might mention your limit to users so that they aren’t surprised when their mail bounces. We recommend a fairly high limit (such as 50MB) anyway, since some legitimate mail is huge.

The ConnectionRateThrottle option, which limits the number of permitted connections per second, can slow things down a bit. Finally, setting MaxRcptsPerMessage, which controls the maximum number of recipients allowed on a single message, may also help.

sendmail has always been able to refuse connections (option REFUSE_LA) or queue email (QUEUE_LA) according to the system load average. A variation, DELAY_LA, keeps the mail flowing, but at a reduced rate.

In spite of all these protections for your mail system, someone mail bombing you will still interfere with legitimate mail. Mail bombing can be quite nasty.

**TLS: Transport Layer Security**
TLS, a encryption/authentication system, is specified in RFC3207. It is implemented in sendmail as an extension to SMTP called STARTTLS.

Strong authentication can replace a hostname or IP address as the authorization token for relaying mail or for accepting a connection from a host in the first place. An entry such as

    TLS_Srv:secure.example.com   ENCR:112
    TLS_Clt:laptop.example.com   PERM+VERIFY:112

in the access_db indicates that STARTTLS is in use and that email to the domain secure.example.com must be encrypted with at least 112-bit encryption keys. Email from a host in the laptop.example.com domain should be accepted only if the client has authenticated itself.

Although STARTTLS provides strong encryption, note that its protection covers only the journey to the “next hop” MTA. Once the message arrives at the next hop, it might be forwarded to another MTA that does not use a secure transport method. If you have control of all possible MTAs in the path, you can create a secure mail transport network. If not, you will need to rely on a UA-based encryption package (such as PGP/GPG) or a centralized email encryption service (see this page).

Greg Shapiro and Claus Assmann of Sendmail, Inc., have stashed some (slightly dated) extra documentation about security and sendmail on the web. It’s available from sendmail.org/~gshapiro and sendmail.org/~ca. The index link in ~ca is especially useful.
**sendmail testing and debugging**

`m4`-based configurations are to some extent pretested. You probably won’t need to do low-level debugging if you use them. But one thing the debugging flags cannot test is your design.

While researching this chapter, we found errors in several of the configuration files and designs that we examined. The errors ranged from invoking a feature without the prerequisite macro (e.g., enabling `masquerade_envelope` without having turned on masquerading with `MASQUERADE_AS`) to total conflict between the design of the `sendmail` configuration and the firewall that controlled whether and under what conditions mail was allowed in.

You cannot design a mail system in a vacuum. You must synchronize it with (or at least not be in conflict with) your DNS MX records and your firewall policy.

**Queue monitoring**

You can use the `mailq` command (which is equivalent to `sendmail -bp`) to view the status of queued messages. Messages are queued while they are being delivered or when delivery has been attempted but has failed.

`mailq` prints a human-readable summary of the files in `/var/spool/mqueue` at any given moment. The output is useful for determining why a message may have been delayed. If it appears that a mail backlog is developing, you can monitor the status of `sendmail`’s attempts to clear the jam.

There are two default queues: one for messages received on port 25 and another for messages received on port 587 (the client submission queue). You can invoke `mailq -Ac` to see the client queue.

Below, some typical output from `mailq` shows three messages waiting to be delivered.

```
$ sudo mailq
/var/spool/mqueue (3 requests)
-----Q-ID----- -Size- ---Q-Time--- --------Sender/Recipient-------
k623gYYk008732 23217 Sat Jul 1 21:42 MAILER-DAEMON
  8BITMIME (Deferred: Connection refused by agribusinessonline.com.)
    <Nmtz@agribusinessonline.com>

k5ULkJhB032374  279 Fri Jun 30 15:46 <randy@atrust.com>
  (Deferred: Name server: k2wireless.com.: host name lookup failed)
    <relder@k2wireless.com>

k5UJDm72923576  2465 Fri Jun 30 13:13 MAILER-DAEMON
  (reply: read error from mx4.level3.com.)
    <lfinist@bbnplanet.com>
```

If you think you understand the situation better than `sendmail` or you just want `sendmail` to try to redeliver the queued messages immediately, you can force a queue run with `sendmail -q`. If you use `sendmail -q -v`, `sendmail` shows the play-by-play results of each delivery attempt,
information that is often useful for debugging. Left to its own devices, \texttt{sendmail} retries delivery every queue run interval (typically every 30 minutes).

\textbf{Logging}

See \textit{Chapter 10} for more information about syslog.

\texttt{sendmail} uses syslog to log error and status messages with the syslog facility “mail” and levels “debug” through “crit”; messages are tagged with the string “sendmail.” You can override the logging string “sendmail” with the \texttt{-L} command-line option; this capability is handy if you are debugging one copy of \texttt{sendmail} while other copies are doing regular email chores.

The \texttt{confLOG_LEVEL} option, specified on the command line or in the config file, determines the severity level that \texttt{sendmail} uses as a threshold for logging. High values of the log level imply low severity levels and cause more info to be logged.

\texttt{Table 18.15} gives an approximate mapping between \texttt{sendmail} log levels and syslog severity levels.

\begin{table}[h]
\centering
\begin{tabular}{|c|c|c|}
\hline
\textbf{L} & \textbf{Syslog levels} & \textbf{L} & \textbf{Syslog levels} \\
\hline
0 & No logging & 4 & notice \\
1 & alert or crit & 5–11 & info \\
2 & crit & \geq 12 & debug \\
3 & err or warning &  &  \\
\hline
\end{tabular}
\caption{sendmail log levels (L) vs. syslog levels}
\end{table}

Recall that a message logged to syslog at a particular level is reported to that level and all those above it. The \texttt{/etc/syslog.conf} or \texttt{/etc/rsyslog.conf} file determines the eventual destination of each message. \texttt{Table 18.16} shows their default locations.

\begin{table}[h]
\centering
\begin{tabular}{|l|l|}
\hline
\textbf{System} & \textbf{Log file location} \\
\hline
Debian & /var/log/mail.log \\
Ubuntu & /var/log/mail.log \\
Red Hat & /var/log/maillog \\
CentOS & /var/log/maillog \\
FreeBSD & /var/log/maillog \\
\hline
\end{tabular}
\caption{Default sendmail log locations}
\end{table}
Several programs can summarize *sendmail* log files, with the end products ranging from simple counts and text tables (*mreport*) to fancy web pages (Yasma). You might need to limit access to this data or at least inform your users that you are collecting it.
18.9 Exim

The Exim mail transport and submission agent was written in 1995 by Philip Hazel of the University of Cambridge and is distributed under the GNU General Public License. The current release, Exim version 4.89, came out in spring 2017. Tons of Exim documentation are available on-line, as are a couple of books by the author of the software.

Googling for Exim questions often seems to lead to old, undated, and sometimes inappropriate materials, so check the official documentation first. A 400+ page specification and configuration document (doc/spec.txt) is included in the distribution. This document is also available from exim.org as a PDF file. It’s the definitive reference work for Exim and is updated religiously with each new release.

There are two cultures with respect to Exim configuration: Debian’s and the rest of the world’s. Debian runs its own set of mailing lists to support users; we do not cover the Debian-specific configuration extensions here.

Exim is like sendmail in that it is implemented as a single process that performs essentially all the ongoing chores associated with email. However, Exim does not carry all sendmail’s historical baggage (support for ancient address formats, needing to get mail to hosts not on the Internet, etc.). Many aspects of Exim’s behavior are specified at compile time, the chief examples being Exim’s database and message store formats.

The workhorses in the Exim system are called routers and transports. Both are included in the general category of “drivers.” Routers decide how messages should be delivered, and transports decide on the mechanics of making deliveries. Routers are an ordered list of things to try, whereas transports are an unordered set of delivery methods.
Exim installation

You can download the latest distribution from exim.org or from your favorite package repository. Refer to the top-level README file and the file src/EDITME, in which you must set installation locations, user IDs, and other compile-time parameters. EDITME is over 1,000 lines long, but it’s mostly comments that lead you through the compilation process; required changes are well labeled. After your edits, save the file as ../Local/Makefile or ../Local/Makefile-osname (if you are building configurations for several different operating systems from the same distribution directory) before you run make.

Here are a few of the important variables (our opinion) and suggested values (Exim developers’ opinion) from the EDITME file. The first five are required, and the rest are recommended.

```
BIN_DIRECTORY=/usr/exim/bin       # Where the exim binary should live
SPool_DIRECTORY=/var/spool/exim   # Mail spool directory
configure_FILE=/usr/exim/configure # Exim’s configuration file
SYSTEM_ALIASES_FILE=/etc/aliases  # Location of aliases file
EXIM_USER=ref:exim                # User to run as after rootly chores

ROUTER_ACCEPT=Yes                # Router drivers to include
ROUTER_DNSLOOKUP=Yes
ROUTER_IPLITERAL=Yes
ROUTER_MANUALROUTE=Yes
ROUTER_QUERYPROGRAM=Yes
ROUTER_REDIRECT=Yes

TRANSPORT_APPENDFILE=Yes         # Transport drivers to include
TRANSPORT_AutoREPLY=Yes
TRANSPORT_PIPE=Yes
TRANSPORT_SMTP=Yes

SUPPORT_MAILDIR=Yes              # Mailbox formats to understand
SUPPORT_MAILSTORE=Yes
SUPPORT_MBX=Yes

LOOKUP_DBM=Yes                   # DB lookup methods to include
LOOKUP_LSEARCH=Yes
LOOKUP_DNSDB=Yes                 # Allow near-arbitrary DNS lookups
USE_DB=Yes
DBMLIB=-1db                      # (from README)
WITH_CONTENT_SCAN=Yes            # Include content scanning via ACLs

EXPERIMENTAL_SPF=Yes             # Include SPF support, needs libspf2
CFLAGS +-I/usr/local/include      # From www.libspf2.org
LDFLAGS +-lspf2

LOG_FILE_PATH=/var/log/exim_%slog # Log files: file, syslog, or both
LOG_FILE_PATH=syslog
LOG_FILE_PATH=syslog:/var/log/exim_%slog
EXICYCLOG_MAX=10                 # Compress/cycle log files, keep 10
```
Routers and transports must be compiled into the code if you intend to use them. In these days of large memories, you might as well leave them all in. Some default paths are certainly nonstandard: for example, the binary in /usr/exim/bin and the PID file in /var/spool/exim. You might want to tweak these values to match your other installed software.

About ten database lookup methods are available, including MySQL, Oracle, and LDAP. If you include LDAP, you must specify the LDAP_LIB_TYPE variable to tell Exim which LDAP library you are using. You may also need to specify the path to LDAP include files and libraries.

The EDITME file does a good job of telling you about any dependencies your database choices might entail. Any entries above that have “(from README)” in their comment line were not listed in src/EDITME but rather in the README.

EDITME has many additional security options that you might want to include, such as support for SMTP AUTH, TLS, PAM, and options for controlling file ownerships and permissions. You can disable certain Exim options at compile time to limit the damage a hacker might cause if the software is compromised.

It’s advisable to read the entire EDITME file before you complete the installation. It gives you a good feel for what you can control at run time through the configuration file. The top-level README file has lots of detail about OS-specific quirks that you might need to add to the EDITME file as well.

Once you have modified EDITME and installed it as Local/Makefile, run make at the top of the distribution tree followed by sudo make install. The next step is to test your shiny new exim binary and see if it delivers mail as expected. The doc/spec.txt file contains good testing documentation.

Once you are satisfied that Exim is working properly, link /usr/sbin/sendmail to exim so that Exim can emulate the traditional command-line interface to the mail system used by many user agents. You must also arrange for exim to be started at boot time.
Exim startup

On a mail hub machine, **exim** typically starts at boot time in daemon mode and runs continuously, listening on port 25 and accepting messages through SMTP. See *Chapter 2, Booting and System Management Daemons*, for startup details for your operating system.

Like **sendmail**, Exim can wear several hats, and if started with specific flags or alternative command names, it performs different functions. Exim’s mode flags are similar to those understood by **sendmail** because **exim** works hard to maintain compatibility when called by user agents and other tools. **Table 18.17** lists a few common flags.

**Table 18.17: Common exim command-line flags**

<table>
<thead>
<tr>
<th>Flag</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>-bd</td>
<td>Runs in daemon mode and listens for connections on port 25</td>
</tr>
<tr>
<td>-bf or -bF</td>
<td>Runs in user or system filter test mode</td>
</tr>
<tr>
<td>-bi</td>
<td>Rebuilds hashed aliases (same as newaliases)</td>
</tr>
<tr>
<td>-bp</td>
<td>Prints the mail queue (same as mailq)</td>
</tr>
<tr>
<td>-bt</td>
<td>Enters address test mode</td>
</tr>
<tr>
<td>-bV</td>
<td>Checks for syntax errors in the configuration file</td>
</tr>
<tr>
<td>-d+-category</td>
<td>Runs in debug mode, flexible category-based configuration</td>
</tr>
<tr>
<td>-q</td>
<td>Starts a queue runner (same as runq)</td>
</tr>
</tbody>
</table>

Any errors in the config file that can be detected at parse time are caught by **exim -bV**, but some errors can only be caught at run time. Misplaced braces are a common mistake.

The **exim** man page gives lots of detail on all the nooks and crannies of **exim**’s command-line flags and options, including extensive debugging information.
Exim utilities

The Exim distribution includes a bunch of utilities to help you monitor, debug, and sanity-check your installation. Below is the current list along with a brief description of each. See the documentation from the distribution for more detail.

- **exicyclog** – rotates log files
- **exigrep** – searches the main log
- **exilog** – visualizes log files across multiple servers
- **exim_checkaccess** – checks address acceptance from a given IP address
- **exim_dbmbuild** – builds a DBM file
- **exim_dumpdb** – dumps a hints database
- **exim_fixdb** – patches a hints database
- **exim_lock** – locks a mailbox file
- **exim_tidydb** – cleans up a hints database
- **eximstats** – extracts statistics from the log
- **exinext** – extracts retry information
- **exipick** – selects messages according to various criteria
- **exiqgrep** – searches the queue
- **exiqsumm** – summarizes the queue
- **exiwhat** – lists what Exim processes are doing

Another utility that is part of the Exim suite is **eximon**, an X Windows application that displays Exim’s state, the state of Exim’s queue, and the tail of the log file. As with the main distribution, you build it by editing a well-commented EDITME file in the **exim_monitor** directory and running **make**. However, in the case of **eximon** the defaults are usually fine, so you should not have to do much configuration to build the application. Some configuration and queue management can be done from the **eximon** GUI as well.
Exim configuration language

The Exim configuration language (or more accurately, languages: one for filters, one for regular expressions, etc.) feels a bit like the ancient (1970s) language Forth. When first reading an Exim configuration, you might find it hard to distinguish between keywords and option names (which are fixed by Exim) and variable names (which are defined by sysadmins through configuration statements).

Although Exim is advertised as being easy to configure and is extensively documented, there can be quite a learning curve for new users. The section “How Exim receives and delivers mail” in the specification document is essential reading for newcomers. It gives a good feel for the underlying concepts of the system.

When assigned a value, the Exim language’s predefined options sometimes cause an action. The values of about 120 predefined variables may also change in response to an action. These variables can be included in conditional statements.

The language for evaluating if statements and the like may remind you of the reverse Polish notation used during the heyday of Hewlett-Packard calculators. Let’s look at a simple example. In the line

```
 acl_smtp_rcpt = ${if ={25}{$interface_port} \\
                  {acl_check_rcpt} {acl_check_rcpt_submit} }
```

the acl_smtp_rcpt option, when set, causes an ACL to be implemented for each recipient (SMTP RCPT command) in the SMTP exchange. The value assigned to this option is either acl_check_rcpt or acl_check_rcpt_submit, depending on whether or not the Exim variable $interface_port has value 25.

We do not detail the Exim configuration language in this chapter, but refer you instead to the extensive documentation. In particular, pay close attention to the string expansion section of the Exim specification.
Exim configuration file

Exim’s run-time behavior is controlled by a single configuration file, usually called `/usr/exim/configure`. Its name is one of the required variables specified in the `EDITME` file and compiled into the binary.

The supplied default configuration file, `src/configure.default`, is well commented and is a good starting place for sites just getting set up with Exim. In fact, we recommend that you don’t stray too far from it until you thoroughly understand the Exim paradigm and need to elaborate on the default configuration for a specific purpose. Exim works hard to support common situations and has sensible defaults.

It’s also helpful to stick with the variable names used in the default config file. These naming conventions are assumed by folks on the exim-users mailing list. Those people are also a good resource to consult regarding your configuration questions.

`exim` prints a message to stderr and exits if you have a syntax error in your configuration file. It doesn’t catch all syntax errors immediately, however, because it does not expand variables until it needs to.

The order of entries in the configuration file is not quite arbitrary: the global configuration options section must be first and must exist. All other sections are optional and can appear in any order.

Possible sections include

- Global configuration options (mandatory)
- `acl` – access control lists that filter addresses and messages
- `authenticators` – for SMTP AUTH or TLS authentication
- `routers` – ordered sequence to determine where a message should go
- `transports` – definitions of the drivers that do the actual delivery
- `retry` – policy settings for dealing with problem messages
- `rewrite` – global address rewriting rules
- `local_scan` – a hook for fancy flexibility

Each section except the first starts with a `begin section-name` statement—for example, `begin acl`. There is no `end section-name` statement; the end is signaled by the next section’s `begin` statement. Indentation to show subordination makes the config file easier to read for humans, but it is not meaningful to Exim.
Some configuration statements name objects that will later be used to control the flow of messages. Those names must begin with a letter and contain only letters, numbers, and the underscore character. If the first non-whitespace character on a line is #, the rest of the line is treated as a comment. Note that this means you cannot put a comment on the same line as a statement; it will not be recognized as a comment because the first character is not #.

Exim lets you include files anywhere in the configuration file. Two forms of include are used:

```
.include absolute-path
.include_if_exists absolute-path
```

The first form generates an error if the file does not exist. Although include files keep your config file tidy, they are read several times during the life of a message, so it might be best just to include their contents directly into your configuration.
**Global options**

Lots of stuff is specified in the global options section, including operating parameters (limits, sizes, timeouts, properties of the mail server on this host), list definitions (local hosts, local hosts to relay for, remote domains to relay for), and macros (hostname, contact, location, error messages, SMTP banner).

**Options**

Options are set with the basic syntax

```
option_name = value[s]
```

where the *values* can be Booleans, strings, integers, decimal numbers, or time intervals. Multivalued options are allowed, in which case the various values are separated by colons.

Use of the colon as a value separator presents a problem when you express IPv6 addresses, which use colons as part of the address. You can escape the colons by doubling them, but the easiest and most readable fix is to redefine the separator character with the `<` character as you assign values to the option. For example, both of the following two lines set the value of the `localhost_interfaces` option, which contains the IPv4 and IPv6 localhost addresses:

```
local_interfaces = 127.0.0.1 : :::1
local_interfaces = <; 127.0.0.1 ; ::1
```

The second form, in which the semicolon has been defined as the separator, is more readable and less fragile.

There are a zillion options—more than 500 in the options index of the documentation. And we said *sendmail* was complicated! Most options have sensible defaults, and all have descriptive names. It’s handy to have a copy of the `doc/spec.txt` file from the distribution in your favorite text editor when you are researching a new option. We don’t cover all the options below, just the ones that occur in our example configuration bits.

**Lists**

Exim has four kinds of lists, introduced by the keywords `hostlist`, `domainlist`, `addresslist`, and `localpartslist`. Here are two examples that use `hostlist`:

```
hostlist my_relay_list = 192.168.1.0/24 : myfriend.example.com
hostlist my_relay_list = /usr/local/exim/relay_hosts.txt
```

Members can be listed in-line or taken from a file. If in-line, they are separated by colons. There can be up to 16 named lists of each type. In the in-line example above, we included all machines on a local /24 network and a specific hostname.
The symbol @ can be a member of a list; it means the name of the local host and helps you write a single generic configuration file that works for most nonhub machines at your site. The notation @[] is also useful and means all IP addresses on which Exim is listening; that is, all the IP addresses of the local host.

Lists can include references to other lists and the ! character to indicate negation. Lists that include references to variables (e.g., $variable_name) make processing slower because Exim cannot cache the results of evaluating the list, which it otherwise does by default.

To reference a list, just put + in front of its name to match members of the list or !+ to match nonmembers; for example, +my_relay_list. Omit space between the + sign and the name of the list.

Macros

You can use macros to define parameters, error messages, etc. The parsing is primitive, so you cannot define a macro whose name is a subset of another macro without unpredictable results.

The syntax is

MACRO_NAME = rest of the line

For example, the first of the following lines defines a macro named ALIAS_QUERY that looks up a user’s alias entry in a MySQL database. The second line shows the use of the macro to perform an actual lookup, with the result being stored in the variable called data.

```
ALIAS_QUERY = \\
    select mailbox from user where login = '${quote_mysql:$local_part}';
data = ${lookup mysql[ALIAS_QUERY]}
```

Macro names are not required to be all caps, but they must begin with a capital letter. However, the all-caps convention aids clarity. The configuration file can include ifdefs that evaluate a macro and use it to determine whether or not to include a portion of the config file. Every imaginable form of ifdef is supported; they all begin with a dot.
Access control lists (ACLs)

Access control lists filter the addresses of incoming messages and either accept or deny them. Exim divides incoming addresses into a local part that represents the user and a domain part that is the recipient’s domain.

ACLs can be applied at any of the various stages of an SMTP conversation: HELO, MAIL, RCPT, DATA, etc. Typically, an ACL enforces strict adherence to the SMTP protocol at the HELO stage, checks the sender and the sender’s domain at the MAIL stage, checks the recipients at the RCPT stage, and scans the message content at the DATA stage.

A slew of options named acl_smtp_command specify which ACL should be applied after each command in the SMTP protocol. For example, the acl_smtp_rcpt option directs the ACL to run on each address that is a recipient of the message. Another commonly used checkpoint is acl_smtp_data, which checks the ACL against the message after it has been received, for example, to scan content.

You can define ACLs in the acl section of the config file, in a file that is referenced by the acl_smtp_command option or in-line when the option is defined.

A sample ACL called my_acl_check_rcpt is defined below. We would invoke it by assigning its name to the acl_smtp_rcpt option in the global options section of the config file. (If this ACL denies an address at the level of the RCPT command, the sending server should give up and not try the address again.)

This is a long ACL specification, so we break it up into digestible pieces that we can decode individually.

The first portion:

```
begin acl
  my_acl_check_rcpt:
    accept   hosts = :
      control = dkim_disable_verify
```

The default name for this access control list is acl_check_rcpt; you probably should not change its name as we did here. We used a nonstandard name simply to emphasize that the name is something you specify, not a keyword that’s special to Exim.

The first accept line, containing just a colon, is an empty list. The empty list of remote hosts matches cases in which a local MUA submitted a message on the MTA’s standard input. If the address being tested meets this condition, the ACL accepts the address and disables DKIM signature validation, which is turned on by default. If the address does not match this address clause, control drops through to the next clause in the ACL definition:
deny message = Restricted characters in address
domains = +local_domains
local_parts = ^[.] : ^.*[@%!|]

deny message = Restricted characters in address
domains = !+local_domains
local_parts = ^[./] : ^.*[@%!] : ^.*/\./\./\.

The first deny stanza is intended for messages coming into your local domains. It rejects any address whose local part (the username) starts with a dot or contains the special characters @, %, !, /, or |. The second deny applies to messages being sent out by your users. It, too, disallows certain special characters and sequences in the local parts of addresses, in case your users’ machines have been infected with a virus or other malware. In the past, such addresses have been used by spammers to confuse ACLs or have been associated with other security problems.

In general, if you are intending to use $local_parts (supposedly, the recipient’s username) in a directory path (to store mail or look for a vacation file, for example) be careful that your ACLs have filtered out any special characters that could cause unwanted behavior. (The example looks for the sequence /../, which could be problematic if the username were inserted into a path.)

accept local_parts = postmaster
domains = +local_domains

This accept stanza guarantees that mail to postmaster always gets through if it’s sent to a local domain, and that can help with debugging.

require verify = sender

The require line checks to see if a bounce message can be returned; however, it checks only the sender’s domain. (require means “deny if not matched.”) If the sender’s username has been forged, a bounce message could still fail; that is, the bounce message itself could bounce. You can add more extensive checking here by calling another program, but some sites consider such callouts abusive and might add your mail server to a blacklist or bad-reputation list.

accept hosts = +relay_from_hosts
control = submission
control = dkim_disable_verify

The above accept stanza checks for hosts that are allowed to relay through this host, namely, local hosts that are submitting mail into the system. The control line specifies that Exim should act as a mail submission agent and fix up any header deficiencies as the message arrives from the user agent. The recipient’s address is not checked because many user agents are confused by error returns. (This part of the configuration is appropriate only for local machines that relay to a smart host, not for any external domains you might be willing to relay for.) DKIM verification is disabled because these messages are outbound from your users or relay friends.
The last `accept` stanza deals with local hosts that authenticate through SMTP AUTH. Once again, these messages are treated as submissions from user agents.

```
accept authenticated = *
control = submission
control = dkim_disable_verify
```

Here, we check the destination domain to which the message is headed and require that it be either in our list of `local_domains` or in our list of domains to which we allow relaying, `relay_to_domains`. (These domain lists are defined outside the context of the ACL.) Any destinations not in one of those lists are refused with a customized error message.

```
require message = Relay not permitted
domains = +local_domains : +relay_to_domains

require verify = recipient
```

Finally, given that all previous requirements have been met but that no more-specific `accept` or `deny` rule has been triggered, we verify the recipient and accept the message. Most Internet messages to local users fall into this category.

We haven’t included any blacklist scanning in the example above. To access a blacklist, use one of the examples in the default config file or something like this:

```
deny condition = @{$if isip4{$sender_host_address}}
!authenticated = *
!hosts = +my_whitelist_ips
!dnslists = list.dnswl.org
domains = +local_domains
verify = recipient
message = You are on RBL $dnslist_domain: $dnslist_text
dnslists = zen.spamhaus.org
logwrite = Blacklisted sender [$sender_host_address] \n$dnslist_domain: $dnslist_text
```

Translated to English, the code specifies that if a message matches all of the following criteria, it is rejected with a custom error message and logged (also with a custom message):

- It’s from an IPv4 address (some lists don’t handle IPv6 correctly).
- It’s not associated with an authenticated SMTP session.
- It’s from a sender not in the local whitelist.
- It’s from a sender not in the global (Internet) whitelist.
• It’s addressed to a valid local recipient.

• The sending host is on the zen.spamhaus.org blacklist.

The variables `dnslist_text` and `dnslist_domain` are set by the assignment to `dnslists`, which triggers the blacklist lookup. This `deny` clause could be placed right after your checks for unusual characters in addresses.

Here’s another example ACL that rejects mail if the remote side does not say HELO properly:

```plaintext
acl_check_mail:
  deny     message = 503 Bad command - must send HELO/EHLO first
            condition = ${if !def:sender_helo_name}
  accept
```

Exim solves the early talker problem (a more specific case of “not saying HELO properly”) with the `smtp_enforce_sync` option, which is turned on by default.
Content scanning at ACL time

Exim supports powerful content scanning at several points in a message’s traversal of the mail system: at ACL time (after the SMTP DATA command); at delivery time through the transport_filter option; or with a local_scan function after all ACL checks have been completed. You must compile support for content scanning into Exim by setting the WITH_CONTENT_SCAN variable in the EDITME file; it is commented out by default. This option endows ACLs with extra power and flexibility and adds two new configuration options: spamd_address and av_scanner.

Scanning at ACL time allows a message to be rejected in-line with the MTA’s conversation with the sending host. The message is never accepted for delivery, so it need not be bounced. This way of rejecting the message is nice because it avoids backscatter spam caused by bounce messages to forged sender addresses.
Authenticators

Authenticators are drivers that interact with the SMTP AUTH command’s challenge-and-response sequence and identify an authentication mechanism acceptable to both client and server. Exim supports the following mechanisms:

- **AUTH_CRAM_MD5** (RFC2195)
- **AUTH_PLAINTEXT**, which includes both PLAIN and LOGIN
- **AUTH_SPA**, which supports Microsoft’s Secure Password Authentication

If Exim is receiving email, it is acting as an SMTP AUTH server. If it is sending mail, it is a client. Options that appear in the definitions of authenticator instances are tagged with a prefix of either `server_` or `client_` to allow for different configurations depending on the role Exim is playing.

Authenticators are used in access control lists, as in the following clause in the ACL example from this page:

```
deny !authenticated = *
```

Below is an example that shows both the client-side and server-side LOGIN mechanisms. This simple example uses a fixed username and password, which is OK for small sites but probably inadvisable for larger installations.

```
begin authenticators

my_client_fixed_login:
  driver = plaintext
  public_name = LOGIN
  client_send = : myusername : mypasswd

my_server_fixed_login:
  driver = plaintext
  public_name = LOGIN
  server_advertise_condition = ${if def:tis_cipher}
  server_prompts = User Name : Password
  server_condition = ${if and {{eq($auth1}{username}}} \ 
    {eq($auth2){mypassword}}}})
  server_set_id = $auth1
```

Authentication data can come from many sources: LDAP, PAM, `/etc/passwd`, etc. The `server_advertise_condition` clause above prevents mail clients from sending passwords in the clear by requiring TLS security (through STARTTLS or SSL) on connection. If you want the same behavior when Exim acts as the client system, use the `client_condition` option in the client clause, too, again with `tis_cipher`. 
Refer to the Exim documentation for details of all possible authentication options and for examples.
Routers

Routers work on recipient email addresses, either by rewriting them or by assigning them to a transport and sending them on their way. A particular router can have multiple instances, each with different options.

You specify a sequence of routers. A message starts with the first router and progresses through the list until the message is either accepted or rejected. The accepting router typically hands the message to a transport driver. Routers handle both incoming and outgoing messages. They feel a bit like subroutines in a programming language.

A router can return any of the dispositions shown in Table 18.18 for a message.

<table>
<thead>
<tr>
<th>Status</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>accept</td>
<td>The router accepts the address and hands it to a transport driver.</td>
</tr>
<tr>
<td>pass</td>
<td>The router can’t handle the address; go on to the next router.</td>
</tr>
<tr>
<td>decline</td>
<td>The router chooses not to handle the address; next router, please!</td>
</tr>
<tr>
<td>fail</td>
<td>The address is invalid; the router queues it for a bounce message.</td>
</tr>
<tr>
<td>defer</td>
<td>The message is left in the queue to be dealt with later.</td>
</tr>
<tr>
<td>error</td>
<td>There is an error in the router specification; the message is deferred.</td>
</tr>
</tbody>
</table>

If a message receives a pass or decline from all the routers in the sequence, it is unroutable. Exim bounces or rejects such messages, depending on the context.

If a message meets the preconditions for a router and the router ends with a no_more statement, then that message will not be presented to any additional routers, regardless of its disposition by the current router. For example, if your remote SMTP router has the precondition domains = !+local_domains and has no_more set, then only messages to local users (that is, those that would fail the domains precondition) will continue to the next router in the sequence.

Routers have many possible options; some common examples are preconditions, acceptance or failure conditions, error messages to return, and transport drivers to use.

The next few sections detail the routers called accept, dnslookup, manualroute, and redirect. The example configuration snippets assume that Exim is running on a local machine in the example.com domain. They’re all pretty straightforward; refer to the documentation if you want to use some of the fancier routers.

The accept router

The accept router labels an address as OK and passes the associated message to a transport driver. Below are examples of accept router instances called localusers, for delivering local mail, and
save_to_file, for appending to an archive.

localusers:
  driver = accept
domains = example.com
check_local_user
transport = my_local_delivery

save_to_file:
  driver = accept
domains = dialup.example.com
transport = batchsmtp_appendfile

The localusers router instance checks that the domain part of the destination address is example.com and that the local part of the address is the login name of a local user. If both conditions are met, the router hands the message to the transport driver instance called my_local_delivery, which is defined in the transports section. The save_to_file instance is designed for dial-up users; it appends the message to a file specified in the batchsmtp_appendfile transport definition.

The dnslookup router

The dnslookup router typically handles outgoing messages. It looks up the MX record of the recipient’s domain and hands the message to an SMTP transport driver for delivery. Here is an instance called remoteusers:

remoteusers:
  driver = dnslookup
domains = !+example.com
transport = my_remote_delivery

See this page for more information about RFC1918 private address spaces.

The dnslookup code looks up the MX records for the addressee. If no MX records exist, it tries the A record. A common extension to this router instance prohibits delivery to certain IP addresses; a prime example is the RFC1918 private addresses that cannot be routed on the Internet. See the ignore_target_hosts option for more information.

The manualroute router

The flexible manualroute driver can pretty much route email in whatever way you want. The routing information can be a table of rules that match by recipient domain (route_list) or a single rule that applies to all domains (route_data).

Below are two examples of manualroute instances. The first example implements the “smart host” concept, in which all outgoing nonlocal mail is sent to a central (“smart”) host for processing.
This instance is called `smarthost` and applies to all recipients’ domains that are not (the ! character) in the `local_domains` list.

```
smarthost:
  driver = manualroute
  domains = !+local_domains
  transport = remote_smtp
  route_data = smarthost.example.com
```

The router instance below, `firewall`, speaks SMTP to send incoming messages to hosts inside the firewall (perhaps after scanning them for spam and viruses). It looks up the routing data for each recipient domain in a DBM database that contains the names of local hosts.

```
firewall:
  driver = manualroute
  transport = remote-smtp
  route_data = ${lookup{$domain} dbm {/internal/host/routes}}
```

**The redirect router**

The `redirect` driver does address rewriting, such as that called for in the system-wide `aliases` file or in a user’s `~/.forward` file. It usually does not assign the rewritten address to a transport; that task is left to other routers in the chain.

The first instance shown below, `system_aliases`, looks up aliases with a linear search (`lsearch`) of the `/etc/aliases` file. That’s fine for a small `aliases` file, but if yours is huge, replace that linear search with a database lookup. The second instance, `user_forward`, first verifies that mail is addressed to a local user, then checks that user’s `./forward` file.

```
system_aliases:
  driver = redirect
  data = ${lookup{$local_part} lsearch {/etc/aliases}}

user_forward:
  driver = redirect
  check_local_user
  file = $home/.forward
  no_verify
```

The `check_local_user` option ensures that the recipient is a valid local user. The `no_verify` says not to verify the validity of the address to which the forward file redirects the message; just ship it.

**Per-user filtering through .forward files**

Exim not only allows forwarding through `.forward` files but also allows filtering. It supports its own filtering system as well as the Sieve filtering that is being standardized by the IETF. If the first line of a user’s `.forward` file is
then the subsequent filtering commands (there are about 15 of them) can determine where the message should be delivered. Filtering does not actually deliver messages—it just meddles with the destination. For example:

```bash
#Exim filter
if $header_subject: contains sysadmin
then
    save $home/mail/sysadmin
endif
```

Lots of options are available to control what users can and cannot do in their .forward files. The option names begin with `forbid_` or `allow_`. They’re important because they can prevent users from running shells, loading libraries into binaries, or accessing the embedded Perl interpreter when they shouldn’t. Check for new `forbid_*` options when you upgrade to be sure your users can’t get too fancy in their .forward files.
Transports

Routers decide where messages should go, and transports actually take them there. Local transports typically append to a file, pipe to a local program, or speak the LMTP protocol to an IMAP server. Remote transports speak SMTP to their counterparts across the Internet.

There are five Exim transports: appendfile, lmtp, smtp, autoreply, and pipe; we detail appendfile and smtp. The autoreply transport is typically used to send vacation messages, and the pipe transport hands messages as input to a command through a UNIX pipe. As with routers, you must define instances of transports, and it's OK to have multiple instances of the same type of transport. Order is significant for routers, but not for transports.

The appendfile transport

The appendfile driver stores messages in mbox, mbx, Maildir, or mailstore format in a specified file or directory. You must have included the appropriate mailbox formats when you compiled Exim; they are commented out of the EDITME file by default.

The following example defines the my_local_delivery transport (an instance of the appendfile transport) referred to in the localusers router instance definition on this page.

my_local_delivery:
   driver = appendfile
   file = /var/mail/$local_part
   delivery_date_add
   envelope_to_add
   return_path_add
   group = mail
   mode = 0660

The various *_add lines add headers to the message. The group and mode clauses ensure that the transport agent can write to the file.

The smtp transport

The smtp transport is the workhorse of any mail system. Here, we define two instances, one for the standard SMTP port (25) and one for the mail submission port (587).

my_remote_delivery:
   driver = smtp

my_remote_delivery_port587:
   driver = smtp
   port = 587
   headers_add = X-processed-by: MACRO_HEADER port 587
The second instance, `my_remote_delivery_port587`, specifies the port and also a header to be added to the message that includes an indication of the outgoing port. `MACRO_HEADER` would be defined elsewhere in the configuration file.
Retry configuration

The retry section of the configuration file must exist or Exim will never attempt redelivery of messages that could not be delivered on the first attempt. You can specify three time intervals, each less frequent than the previous one. After the last interval has expired, messages bounce back to the sender as undeliverable. retry statements understand the suffixes m, h, d, and w to indicate minutes, hours, days, and weeks. You can specify different intervals for different hosts or domains.

Here’s what a retry section looks like:

```
begin retry
  *   F, 2h, 15m;  F, 24h, 1h;  F, 4d, 6h
```

This example means, “For any domain, an address that fails temporarily should be retried every 15 minutes for 2 hours, then every hour for the next 24 hours, then every 6 hours for 4 days, and finally, bounced as undeliverable.”
**Rewriting configuration**

The rewriting section of the configuration file starts with `begin rewrite`. It’s used to fix up addresses, not to reroute messages. For example, you could use it on your outgoing addresses:

- To make mail appear to be from your domain, not from individual hosts
- To map usernames to a standard format such as First.Last

Do not apply rewriting to addresses in incoming mail.
Local scan function

To further customize Exim, for example, to filter for the latest and greatest virus, you could write a C function that does your scanning and install it in the `local_scan` section of the config file. Refer to the Exim documentation for details and examples of how to do this.
Logging

Exim by default writes three different log files: a main log, a reject log, and a panic log. Each log entry includes the time the message was written. You specify the location of the log files in the EDITME file (before building Exim) or in the run-time config file in the value of the log_file_path option. By default, logs are kept in the /var/spool/exim/log directory.

The log_file_path option accepts up to two colon-separated values. Each value must be either the keyword syslog or an absolute path with a %s embedded where the names main, reject, and panic can be substituted. For example,

log_file_path = syslog : /var/log/exim_%s

would log both to syslog (with facility “mail”) and to the separate files exim_main, exim_reject, and exim_panic in the /var/log directory. Exim submits the main log entries to syslog at priority info, the reject entries at priority notice, and the panic entries at priority alert.

The main log contains one line for the arrival and delivery of each message. It can be summarized by the Perl script eximstats, which is included in the Exim distribution.

The reject log records information about messages that have been rejected for policy reasons: malware, spam, etc. It includes the summary line for the message from the main log and also the original headers of the message that was rejected. If you change your policies, check the reject log to make sure that all is still well.

The panic log is for serious errors in the software; exim writes here just before it gives up. The panic log should not exist in the absence of problems. Ask cron to check it for you and if it exists, fix the problem that caused the panic and then delete the file. exim will re-create it when the next panic-worthy situation arises.

When debugging, you can increase the amount and type of data logged. Invoke the log_selector option. For example:

log_selector = +smtp_connection +smtp_incomplete_transaction +...

The logging categories that can be included or excluded by the log_selector mechanism are listed in the Exim specification, in the section called “Log files” toward the end. About 35 categories are defined, including +all, which will really fill your disks!

exim also keeps a temporary log for each message it handles. It is named with the message ID and lives in /var/spool/exim/msglog. If you are having trouble with a particular destination, check there.
Debugging

Exim has powerful debugging aids. You can configure the amount of information you want to see about each potential debugging topic. `exim -d` tells `exim` to go into debugging mode, in which it stays in the foreground and does not detach from the terminal. You can add specific debugging categories to `-d` with a `+` or `-` in front of them to verbosify or eliminate a category. For example, `-d+expand+acl` requests regular debugging output plus extra details regarding string expansions and ACL interpretation. (These two categories are common problem spots.) You can tune more than 30 categories of debugging information; see the man page for a list.

A common technique when debugging mail systems is to start the MTA on a nonstandard port and then talk to it through `telnet`. For example, to start `exim` in daemon mode, listening on port 26, with debugging info turned on, run

```bash
$ sudo exim -d -oX 26 -bd
```

You can then `telnet` to port 26 and type SMTP commands in an attempt to reproduce the problem you are debugging.

Alternatively, you can have `swaks` do your SMTP talking for you. It’s a Perl script that makes SMTP debugging faster and easier. `swaks --help` gets you some documentation, and [jetmore.org/john/code/swaks](http://jetmore.org/john/code/swaks) supplies complete details.

If your log files show timeouts of around 30 seconds, that’s suggestive of a DNS issue.
18.10 Postfix

Postfix is another popular alternative to sendmail. Wietse Venema started the Postfix project when he spent a sabbatical year at IBM’s T. J. Watson Research Center in 1996, and he is still actively developing it. Postfix’s design goals included not only security (first and foremost!), but also an open source distribution policy, speedy performance, robustness, and flexibility. All major Linux distributions include Postfix, and since version 10.3, macOS has shipped Postfix instead of sendmail as its default mail system.

See this page for more information about regular expressions.

The most important things to know about Postfix are, first, that it works almost out of the box (the simplest config files are only a line or two long), and second, that it leverages regular expression maps to filter email effectively, especially in conjunction with the PCRE (Perl-Compatible Regular Expression) library. Postfix is compatible with sendmail in the sense that Postfix’s aliases and .forward files have the same format and semantics as those of sendmail.

Postfix speaks ESMTP. Virtual domains and spam filtering are both supported. For address rewriting, Postfix relies on table lookups from flat files, Berkeley DB, DBM, LDAP, NetInfo, or SQL databases.
Postfix architecture

Postfix comprises several small, cooperating programs that send network messages, receive messages, deliver email locally, etc. Communication among them is performed through local domain sockets or FIFOs. This architecture is quite different from that of sendmail and Exim, wherein a single large program does most of the work.

The master program starts and monitors all Postfix processes. Its configuration file, master.cf, lists the subsidiary programs along with information about how they should be started. The default values set in that file cover most needs; in general, no tweaking is necessary. One common change is to comment out a program, for example, smtpd, when a client should not listen on the SMTP port.

The most important server programs involved in the delivery of email are shown in Exhibit B.

Exhibit B: Postfix server programs

Receiving mail

smtpd receives mail entering the system through SMTP. It also verifies that the connecting clients are authorized to send the mail they are trying to deliver. When email is sent locally through the /usr/lib/sendmail compatibility program, a file is written to the /var/spool/postfix/maildrop directory. That directory is periodically scanned by the pickup program, which processes any new files it finds.

All incoming email passes through cleanup, which adds missing headers and rewrites addresses according to the canonical and virtual maps. Before inserting mail into the incoming queue, cleanup passes it through trivial-rewrite, which does minor fixing of the addresses, such as appending a mail domain to addresses that are not fully qualified.

Managing mail-waiting queues

qmgr manages five queues that contain mail waiting to be delivered:
• incoming – mail that is arriving
• active – mail that is being delivered
• deferred – mail for which delivery has failed in the past
• hold – mail blocked in the queue by the administrator
• corrupt – mail that can’t be read or parsed

The queue manager generally follows a simple FIFO strategy to select the next message to process, but it also supports a complex preemption algorithm that prefers messages with few recipients over bulk mail.

To avoid overwhelming a receiving host, especially one that has been down, Postfix uses a slow-start algorithm to control how fast it tries to deliver email. Deferred messages are given a try-again time stamp that exponentially backs off so as not to waste resources on undeliverable messages. A status cache of unreachable destinations avoids unnecessary delivery attempts.

Sending mail

$qmgr$, aided by $trivial-rewrite$, decides where a message should be sent. The routing decision made by $trivial-rewrite$ can be overridden through lookup tables ($transport_maps$).

Delivery to remote hosts via the SMTP protocol is performed by the $smtp$ program. $lmtp$ delivers mail with LMTP, the Local Mail Transfer Protocol defined in RFC2033. LMTP is derived from SMTP, but the protocol has been modified so that the mail server is not required to manage a mail queue. This mailer is particularly useful for delivering email to mailbox servers such as the Cyrus IMAP suite.

$local$’s job is to deliver email locally. It resolves addresses in the $aliases$ table and follows instructions found in recipients’ $forward$ files. Messages are forwarded to another address, passed to an external program for processing, or stored in users’ mail folders.

The $virtual$ program delivers email to “virtual mailboxes”; that is, mailboxes that are not related to a local UNIX account but that still represent valid email destinations. Finally, $pipe$ implements delivery through external programs.
Security

Postfix implements security at several levels. Most of the Postfix server programs can run in a chrooted environment. They are separate programs with no parent/child relationship. None of them are setuid. The mail drop directory is group-writable by the postdrop group, to which the postdrop program is setgid.
Postfix commands and documentation

Several command-line utilities permit user interaction with the mail system:

- **postalias** – builds, modifies, and queries alias tables
- **postcat** – prints the contents of queue files
- **postconf** – displays and edits the main configuration file, **main.cf**
- **postfix** – starts and stops the mail system (must be run as root)
- **postmap** – builds, modifies, or queries lookup tables
- **postsuper** – manages mail queues
- **sendmail, mailq, newaliases** – are **sendmail**-compatible replacements

The Postfix distribution includes a set of man pages that describe all the programs and their options. On-line documents at postfix.org explain how to configure and manage various aspects of Postfix. These documents are also included in the Postfix distribution in the **README_FILES** directory.
Postfix configuration

The main.cf file is Postfix’s principal configuration file. The master.cf file configures the server programs. It also defines various lookup tables that are referenced from main.cf and that provide different types of service mappings.

The postconf(5) man page describes every parameter you can set in the main.cf file. There is also a postconf program, so if you just type man postconf, you’ll get the man page for that instead of postconf(5). Use man -s 5 postconf to get the right version.

The Postfix configuration language looks a bit like a series of sh comments and assignment statements. Variables can be referenced in the definition of other variables by being prefixed with a $. Variable definitions are stored just as they appear in the config file; they are not expanded until they are used, and any substitutions occur at that time.

You can create new variables by assigning values to them. Be careful to choose names that do not conflict with existing configuration variables.

All Postfix configuration files, including the lookup tables, consider lines starting with whitespace to be continuation lines. This convention results in readable configuration files, but you must start new lines in column one.

What to put in main.cf

More than 500 parameters can be specified in the main.cf file. However, just a few of them need to be set at an average site. The author of Postfix strongly recommends that only parameters with nondefault values be included in your configuration. That way, if the default value of a parameter changes in the future, your configuration will automatically adopt the new value.

The sample main.cf file that comes with the distribution includes many commented-out example parameters, along with some brief documentation. The original version is best left alone as a reference. Start with an empty file for your own configuration so that your settings do not become lost in a sea of comments.

Basic settings

The simplest possible Postfix configuration is an empty file. Surprisingly, this is a perfectly reasonable setup. It results in a mail server that delivers email locally within the same domain as the local hostname and that sends any messages directed to nonlocal addresses directly to the appropriate remote servers.

Null client

Another simple configuration is a “null client”; that is, a system that doesn’t deliver email locally but rather forwards outbound mail to a designated central server. To implement this configuration, you define several parameters, starting with mydomain, which defines the domain
part of the hostname, and myorigin, which is the mail domain appended to unqualified email addresses. If these two parameters are the same, you can write something like this:

```
mydomain = cs.colorado.edu
myorigin = $mydomain
```

Another parameter you should set is mydestination, which specifies the mail domains that are local. If the recipient address of a message has mydestination as its mail domain, the message is delivered through the local program to the corresponding user (assuming that no relevant alias or .forward file is found). If more than one mail domain is included in mydestination, these domains are all considered aliases for the same domain.

For a null client, you want no local delivery, so leave this parameter empty:

```
mydestination =
```

Finally, the relayhost parameter tells Postfix to send all nonlocal messages to a specified host instead of sending them directly to their apparent destinations:

```
relayhost = [mail.cs.colorado.edu]
```

The square brackets tell Postfix to treat the specified string as a hostname (DNS A record) instead of a mail domain name (DNS MX record).

Since null clients should not receive mail from other systems, the last thing to do in a null client configuration is to comment out the smtpd line in the master.cf file. This change prevents Postfix from running smtpd at all. With just these few lines, you’ve defined a fully functional null client!

For a “real” mail server, you’ll need a few more configuration options as well as some mapping tables. We cover these in the next few sections.

**Use of postconf**

postconf is a handy tool that helps you configure Postfix. When run without arguments, it prints all the parameters as they are currently configured. If you name a specific parameter as an argument, postconf prints the value of that parameter. The -d option makes postconf print the defaults instead of the currently configured values. For example:

```
$ postconf mydestination
mydestination =
$ postconf -d mydestination
mydestination = $myhostname, localhost.$mydomain, localhost
```

Another useful option is -n, which tells postconf to print only the parameters that differ from the default. If you ask for help on the Postfix mailing list, that’s the configuration information you should put in your email.
Lookup tables

Many aspects of Postfix’s behavior are shaped through the use of lookup tables, which can map keys to values or implement simple lists. For example, the default setting for the `alias_maps` table is

```
alias_maps = dbm:/etc/mail/aliases
```

Data sources are specified with the notation `type:path`. Multiple values can be separated by commas, spaces, or both. Table 18.19 lists the available data sources; `postconf -m` shows this information as well.

**Table 18.19: Information sources for Postfix lookup tables**

<table>
<thead>
<tr>
<th>Type</th>
<th>Description</th>
</tr>
</thead>
<tbody>
<tr>
<td>cidr</td>
<td>Network addresses in CIDR form</td>
</tr>
<tr>
<td>hash/btree</td>
<td>Berkeley DB hash table or B-tree file (replaces dbm)</td>
</tr>
<tr>
<td>ldap</td>
<td>LDAP directory service</td>
</tr>
<tr>
<td>mysql</td>
<td>MySQL database</td>
</tr>
<tr>
<td>pcre</td>
<td>Perl-compatible regular expressions</td>
</tr>
<tr>
<td>pgsq</td>
<td>PostgreSQL database</td>
</tr>
<tr>
<td>proxy</td>
<td>Access through <a href="https://en.wikipedia.org/wiki/Proxymap">proxymap</a>, e.g., to escape a chroot</td>
</tr>
<tr>
<td>regexp</td>
<td>POSIX regular expressions</td>
</tr>
<tr>
<td>static</td>
<td>Return of value specified as path regardless of the key</td>
</tr>
<tr>
<td>unix</td>
<td>The <code>/etc/passwd</code> and <code>/etc/group</code> files(^a)</td>
</tr>
</tbody>
</table>

\(^a\) `unix:passwd.bname` is the `passwd` file, and `unix:group.bname` is the `group` file.

The `dbm` and `sdbm` types are only for compatibility with the traditional [sendmail](https://en.wikipedia.org/wiki/Sendmail) alias table. Berkeley DB (`hash`) is a more modern implementation; it’s safer and faster. If compatibility is not a problem, then go with

```
alias_database = hash:/etc/mail/aliases
alias_maps = hash:/etc/mail/aliases
```

The `alias_database` specifies the table that is rebuilt by `newaliases` and should correspond to the table that you specify in `alias_maps`. The two parameters are separate because `alias_maps` might include non-DB sources such as `mysql` that never need to be rebuilt.

All DB-class tables (`dbm`, `sdbm`, `hash`, and `btree`) compile a text file to an efficiently searchable binary format. The syntax for these text files is similar to that of the configuration files with respect to comments and continuation lines. Entries are specified as simple key/value pairs separated by whitespace, except for alias tables, which use a colon after the key to retain `sendmail` compatibility. For example, the following lines are appropriate for an alias table:

```
alias_map = dbm:/etc/mail/aliases
```

```
alias = b:b@domain.com:foo@bar.com
```
As another example, here’s an access table for relaying mail from any client with a hostname ending in cs.colorado.edu.

```
.cs.colorado.edu   OK
```

Text files are compiled to their binary formats with the `postmap` command for normal tables and the `postalias` command for alias tables. The table specification (including the type) must be given as the first argument. For example:

```
$ sudo postmap hash:/etc/postfix/access
```

**postmap** can also query values in a lookup table (no match = no output):

```
$ postmap -q blabla hash:/etc/postfix/access
$ postmap -q .cs.colorado.edu hash:/etc/postfix/access
OK
```

**Local delivery**

The **local** program delivers mail to local recipients. It also handles local aliasing. For example, if `mydestination` is set to cs.colorado.edu and email arrives for the recipient evi@cs.colorado.edu, **local** first consults the alias_maps tables and then substitutes any matching entries recursively. If no aliases match, **local** looks for a **.forward** file in user evi’s home directory and follows the instructions in this file if it exists. (The syntax is the same as for the right side of an alias map.) Finally, if no **.forward** file is found, the email is delivered to evi’s local mailbox.

By default, **local** writes to standard **mbox**-format files under `/var/mail`. You can change that behavior with the parameters shown in **Table 18.20**.

**Table 18.20: Parameters for local mailbox delivery (set in main.cf)**

<table>
<thead>
<tr>
<th>Parameter</th>
<th>Description</th>
</tr>
</thead>
<tbody>
<tr>
<td>home_mailbox</td>
<td>Delivers mail to ~user under the specified relative path</td>
</tr>
<tr>
<td>mail_spool_directory</td>
<td>Delivers mail to a central directory that serves all users</td>
</tr>
<tr>
<td>mailbox_command</td>
<td>Delivers mail with an external program, typically <strong>procmail</strong></td>
</tr>
<tr>
<td>mailbox_transport</td>
<td>Delivers mail through a service as defined in <strong>master.cf</strong>&lt;sup&gt;a&lt;/sup&gt;</td>
</tr>
<tr>
<td>recipient_delimiter</td>
<td>Allows extended usernames (see description below)</td>
</tr>
</tbody>
</table>

<sup>a. This option interfaces with mailbox servers such as the Cyrus **imapd**.</sup>

The `mail_spool_directory` and `home_mailbox` options normally generate **mbox**-format mailboxes, but they can also produce **Maildir** mailboxes. To request this behavior, add a slash to the end of the
If `recipient_delimiter` is +, mail addressed to evi+`whatever`@cs.colorado.edu is accepted for delivery to the evi account. With this facility, users can create special-purpose addresses and sort their mail by destination address. Postfix first attempts lookups on the full address, and only if that fails does it strip the extended components and fall back to the base address. Postfix also looks for a corresponding forwarding file, `.forward+whatever`, for further aliasing.
Virtual domains

To host a mail domain on your Postfix mail server, you have three choices:

- List the domain in `mydestination`. Delivery is performed as described above: aliases are expanded and mail is delivered to the corresponding accounts.

- List the domain in the `virtual_alias_domains` parameter. This option gives the domain its own addressing namespace that is independent of the system’s user accounts. All addresses within the domain must be resolvable (through mapping) to real addresses outside of it.

- List the domain in the `virtual_mailbox_domains` parameter. As with the `virtual_alias_domains` option, the domain has its own namespace. All mailboxes must live beneath a specified directory.

List the domain in only one of these three places. Choose carefully, because many configuration elements depend on that choice. We have already reviewed the handling of the `mydestination` method. The other options are discussed below.

Virtual alias domains

If a domain is listed as a value of the `virtual_alias_domains` parameter, mail to that domain is accepted by Postfix and must be forwarded to an actual recipient either on the local machine or elsewhere.

The forwarding for addresses in the virtual domain must be defined in a lookup table included in the `virtual_alias_maps` parameter. Entries in the table have the address in the virtual domain on the left side and the actual destination address on the right. An unqualified name on the right is interpreted as a local username.

Consider the following example from `main.cf`:

```plaintext
myorigin = cs.colorado.edu
mydestination = cs.colorado.edu
virtual_alias_domains = admin.com
virtual_alias_maps = hash:/etc/mail/admin.com/virtual
```

In `/etc/mail/admin.com/virtual` we could then have the lines

```plaintext
postmaster@admin.com   evi, david@admin.com
david@admin.com       david@schweikert.ch
evi@admin.com         evi
```

Mail for `evi@admin.com` would be redirected to `evi@cs.colorado.edu` (myorigin is appended) and would ultimately be delivered to the mailbox of user evi because cs.colorado.edu is included in mydestination.
Definitions can be recursive: the right hand side can contain addresses that are further defined on the left hand side. Note that the right hand side can only be a list of addresses. To execute an external program or to use `:include:` files, redirect the email to an alias, which can then be expanded according to your needs.

To keep everything in one file, set `virtual_alias_domains` to the same lookup table as `virtual_alias_maps` and put a special entry in the table to mark it as a virtual alias domain. In `main.cf`:

```
virtual_alias_domains = $virtual_alias_maps
virtual_alias_maps = hash:/etc/mail/admin.com/virtual
```

In `/etc/mail/admin.com/virtual`:

```
admin.com          notused
postmaster@admin.com  evi, david@admin.com
...
```

The right hand side of the entry for the mail domain (admin.com) is never actually used; admin.com’s existence in the table as an independent entry is enough to make Postfix consider it a virtual alias domain.

**Virtual mailbox domains**

Domains listed under `virtual_mailbox_domains` are similar to local domains, but the list of users and their corresponding mailboxes must be managed independently of the system’s user accounts.

The parameter `virtual_mailbox_maps` points to a table that lists all valid users in the domain. The map format is

```
user@domain       /path/to/mailbox
```

If the path ends with a slash, the mailboxes are stored in Maildir format. The value of `virtual_mailbox_base` is always prefixed to the specified paths.

You often want to alias some of the addresses in the virtual mailbox domain. A `virtual_alias_map` will do that for you. Here is a complete example. In `main.cf`:

```
virtual_mailbox_domains = admin.com
virtual_mailbox_base = /var/mail/virtual
virtual_mailbox_maps = hash:/etc/mail/admin.com/vmailboxes
virtual_alias_maps = hash:/etc/mail/admin.com/valiases
```

```
/etc/mail/admin.com/vmailboxes` might contain entries like these:

```
evi@admin.com       nemeth/evi/
```

```
/etc/mail/admin.com/valiases` might contain:

```
You can use virtual alias maps even on addresses that are not within virtual alias domains. Virtual alias maps let you redirect any address from any domain, independently of the type of the domain (canonical, virtual alias, or virtual mailbox). Since mailbox paths can only be put on the right hand side of the virtual mailbox map, this mechanism is the only way to set up aliases in that domain.
Access control

Mail servers should relay mail for third parties only on behalf of trusted clients. If a mail server forwards mail from unknown clients to other servers, it is a so-called open relay, which is bad. See this page for more details.

Fortunately, Postfix doesn’t act as an open relay by default. In fact, its defaults are quite restrictive; you are more likely to need to liberalize the permissions than to tighten them. Access control for SMTP transactions is configured in Postfix through “access restriction lists.” The parameters shown in Table 18.21 control what should be checked during the different phases of an SMTP session.

Table 18.21: Postfix parameters for SMTP access restriction

<table>
<thead>
<tr>
<th>Parameter</th>
<th>When applied</th>
</tr>
</thead>
<tbody>
<tr>
<td>smtpd_client_restrictions</td>
<td>On connection request</td>
</tr>
<tr>
<td>smtpd_data_restrictions</td>
<td>On DATA command (mail body)</td>
</tr>
<tr>
<td>smtpd_etrn_restrictions</td>
<td>On ETRN command&lt;sup&gt;a&lt;/sup&gt;</td>
</tr>
<tr>
<td>smtpd_helo_restrictions</td>
<td>On HELO/EHLO command (start of the session)</td>
</tr>
<tr>
<td>smtpd_recipient_restrictions</td>
<td>On RCPT TO command (recipient specification)</td>
</tr>
<tr>
<td>smtpd_relay_restrictions</td>
<td>On relay attempt to a third party domain</td>
</tr>
<tr>
<td>smtpd_sender_restrictions</td>
<td>On MAIL FROM command (sender specification)</td>
</tr>
</tbody>
</table>

<sup>a</sup> This is a special command used for resending messages in the queue.

The most important parameter is `smtpd_recipient_restrictions`. That’s because access control is most easily performed when the recipient address is known and can be identified as being local or not. All other parameters in Table 18.21 are empty in the default configuration. The default value is

`smtpd_recipient_restrictions = permit_mynetworks,
                             reject_unauth_destination`

Each of the specified restrictions is tested in turn until a definitive decision about what to do with the mail is reached. Table 18.22 shows the common restrictions.

Table 18.22: Common Postfix access restrictions

<table>
<thead>
<tr>
<th>Restriction</th>
<th>Function</th>
</tr>
</thead>
<tbody>
<tr>
<td>check_client_access</td>
<td>Checks client host address through a lookup table</td>
</tr>
<tr>
<td>check_recipient_access</td>
<td>Checks recipient mail address through a lookup table</td>
</tr>
<tr>
<td>permit_mynetworks</td>
<td>Grants access to addresses listed in mynetworks</td>
</tr>
<tr>
<td>reject_unauth_destination</td>
<td>Rejects mail for nonlocal recipients; no relaying</td>
</tr>
</tbody>
</table>
Everything can be tested in these restrictions, not just specific information like the sender address in the `smtpd_sender_restrictions`. Therefore, for simplicity, you might want to put all the restrictions under a single parameter. Make that `smtpd_recipient_restrictions` because it is the only one that can test everything (except the DATA part).

`smtpd_recipient_restrictions` and `smtpd_relay_restrictions` are where mail relaying is tested. Keep the `reject_unauth_destination` restriction and carefully choose the “permit” restrictions before it.

**Access tables**

Each restriction returns one of the actions shown in Table 18.23. Access tables are used in restrictions such as `check_client_access` and `check_recipient_access` to select an action according to the client host address or recipient address, respectively.

<table>
<thead>
<tr>
<th>Action</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>4nn text</td>
<td>Returns temporary error code 4nn and message text</td>
</tr>
<tr>
<td>5nn text</td>
<td>Returns permanent error code 5nn and message text</td>
</tr>
<tr>
<td>DEFER_IF_PERMIT</td>
<td>If restrictions result in PERMIT, changes it to a temp error</td>
</tr>
<tr>
<td>DEFER_IF_REJECT</td>
<td>If restrictions result in REJECT, changes it to a temp error</td>
</tr>
<tr>
<td>DISCARD</td>
<td>Accepts the message but silently discards it</td>
</tr>
<tr>
<td>DUNNO</td>
<td>Pretends the key was not found; tests further restrictions</td>
</tr>
<tr>
<td>FILTER <code>transport:dest</code></td>
<td>Passes the mail through the filter <code>transport:dest</code></td>
</tr>
<tr>
<td>HOLD</td>
<td>Blocks the mail in the queue</td>
</tr>
<tr>
<td>OK</td>
<td>Accepts the mail</td>
</tr>
<tr>
<td>PREPEND <code>header</code></td>
<td>Adds a header to the message</td>
</tr>
<tr>
<td>REDIRECT <code>addr</code></td>
<td>Forwards the mail to a specified address</td>
</tr>
<tr>
<td>REJECT</td>
<td>Rejects the mail</td>
</tr>
<tr>
<td>WARN <code>message</code></td>
<td>Enters the given warning <code>message</code> in the logs</td>
</tr>
</tbody>
</table>

For example, suppose you wanted to allow relaying for all machines within the `cs.colorado.edu` domain and that you wanted to allow only trusted clients to post to the internal mailing list `newsletter@cs.colorado.edu`. You could implement these policies with the following lines in `main.cf`:

```plaintext
smtpd_recipient_restrictions =
    permit_mynetworks
    check_client_access hash:/etc/postfix/relaying_access
    reject_unauth_destination
    check_recipient_access hash:/etc/postfix/restricted_recipients
```

Note that commas are optional when the list of values for a parameter is specified.

In `/etc/postfix/relaying_access`:
In `/etc/postfix/restricted_recipients`:

```plaintext
newsletter@cs.colorado.edu   REJECT Internal list
```

The text after `REJECT` is an optional string that is sent to the client along with the error code. It tells the sender why the mail was rejected.

**Authentication of clients and encryption**

For users sending mail from home, it is usually easiest to route outgoing mail through the home ISP’s mail server, regardless of the sender address that appears on that mail. Most ISPs trust their direct clients and allow relaying. If this configuration isn’t possible or if you are using a system such as Sender ID or SPF, ensure that mobile users outside your network can be authorized to submit messages to your `smtpd`.

The solution to this problem is to have the SMTP AUTH mechanism authenticate directly at the SMTP level. Postfix must be compiled with support for the SASL library to make this work. You can then configure the feature like this:

```plaintext
smtpd_sasl_auth_enable = yes
smtpd_recipient_restrictions =
    permit_mynetworks
    permit_sasl_authenticated
    ... 
```

You also need to support encrypted connections to avoid sending passwords in clear text. Add lines like the following to `main.cf`:

```plaintext
smtpd_tls_security_level = may
smtpd_tls_auth_only = yes
smtpd_tls_loglevel = 1
smtpd_tls_received_header = yes
smtpd_tls_cert_file = /etc/certs/smtp.pem
smtpd_tls_key_file = $smtpd_tls_cert_file
smtpd_tls_protocols = !SSLv2
```

You need to put a properly signed certificate in `/etc/certs/smtp.pem`. It’s also a good idea to turn on encryption on outgoing SMTP connections:

```plaintext
smtp_tls_security_level = may
smtp_tls_loglevel = 1
```
**Debugging**

When you have a problem with Postfix, first check the log files. The answers to your questions are most likely there; it’s just a question of finding them. Every Postfix program normally issues a log entry for every message it processes. For example, the trail of an outbound message might look like this:

Aug 18 22:41:33 nova postfix/pickup: 0E4A93688: uid=506
   from=<dws@ee.ethz.ch>
Aug 18 22:41:33 nova postfix/cleanup: 0E4A93688:
   message-id= <20040818204132.GA11444@ee.ethz.ch>
Aug 18 22:41:33 nova postfix/qmgr: 0E4A93688:
   from=<dws@ee.ethz.ch>, size=577, nrcpt=1 (queue active)
Aug 18 22:41:33 nova postfix/smtp: 0E4A93688:
   to=<evi@ee.ethz.ch>, relay=tardis.ee.ethz.ch[129.132.2.217],
   delay=0, status=sent (250 Ok: queued as 154D4D930B)
Aug 18 22:41:33 nova postfix/qmgr: 0E4A93688: removed

As you can see, the interesting information is spread over many lines. Note that the identifier 0E4A93688 is common to every line: Postfix assigns a queue ID as soon as a message enters the mail system and never changes it. Therefore, when searching the logs for the history of a message, first concentrate on determining the message’s queue ID. Once you know that, it’s easy to **grep** the logs for all the relevant entries.

Postfix is good at logging helpful messages about problems that it notices. However, it’s sometimes difficult to spot the important lines among the thousands of normal status messages. This is a good place to consider using some of the tools discussed in the section *Management of logs at scale*.

**Looking at the queue**

Another place to look for problems is the mail queue. As in the **sendmail** system, a `mailq` command prints the contents of a queue. You can use it to see if and why a message has become stuck.

Another helpful tool is the **qshape** script that’s shipped with recent versions of Postfix. It shows summary statistics about the contents of a queue. The output looks like this:

```
$ sudo qshape deferred

- T  5 10 20 40 80 160 320 640 1280 1280+
- TOTAL  78  0  0  0  7  3  3  2  12  2  49
- expn.com  34  0  0  0  0  0  0  9  0  0  25
- chinabank.ph  5  0  0  0  1  1  1  2  0  0  0
- prob-helper.biz  3  0  0  0  0  0  0  0  0  0  3
```

**qshape** summarizes the given queue (here, the deferred queue), sorted by recipient domain. The columns report the number of minutes the relevant messages have been in the queue. For
example, you can see that 25 messages bound for expn.com have been in the queue longer than 1,280 minutes. All the destinations in this example are suggestive of messages having been sent from vacation scripts in response to spam.

qshape can also summarize by sender domain with the -s flag.

Soft-bouncing

If soft_bounce is set to yes, Postfix sends temporary error messages whenever it would normally send permanent error messages such as “user unknown” or “relaying denied.” This is a great testing feature; it lets you monitor the disposition of messages after a configuration change without the risk of permanently losing legitimate email. Anything you reject will eventually come back for another try. Don’t forget to turn off this feature when you are done testing or you will have to deal with every rejected message over and over again.
18.11 RECOMMENDED READING

Rather than jumble together the references listed here, we’ve sorted them by MTA and topic.
sendmail references


This book is the definitive tome for sendmail configuration—1,300 pages’ worth. It includes a sysadmin guide as well as a complete reference section. An electronic edition is available, too. The author mix includes two key sendmail developers (Claus and Greg) who enforce technical correctness and add insight to the mix.

Installation instructions and a good description of the configuration file are covered in the Sendmail Installation and Operation Guide, which can be found in the doc/op subdirectory of the sendmail distribution. This document is quite complete, and in conjunction with the README file in the cf directory, gives a good nuts-and-bolts view of the sendmail system.

sendmail.org, sendmail.org/~ca, and sendmail.org/~gshapiro all contain documents, HOWTOs, and tutorials related to sendmail.
Exim references


The Exim specification is the defining document for Exim configuration. It is quite complete and is updated with each new distribution. A text version is included in the file `doc/spec.txt` in the distribution, and a PDF version is available from exim.org. The web site also includes several how-to documents.
Postfix references


This book is the best; it guides you through all the details of Postfix configuration, even for complex environments. The authors are active in the Postfix community and participate regularly on the postfix-users mailing list. The book is unfortunately out of print, but used copies are readily available.
RFCs

RFCs 5321 (updated by 7504) and 5322 (updated by 6854) are the current versions of RFCs 821 and 822. They define the SMTP protocol and the formats of messages and addresses for Internet email. RFCs 6531 and 6532 cover extensions for internationalized email addresses. There are currently almost 90 email-related RFCs, too many to list here. See the general RFC search engine at rfc-editor.org for more.
UNIX and Linux are the predominant platforms for serving web applications. According to data from w3techs.com, 67% of the top one million web sites are served by either Linux or FreeBSD. Above the OS level, open source web server software commands more than 80% of the market.

At scale, web applications do not run on a single system. Instead, a collection of software components distributed through a meshwork of systems cooperate to answer requests as quickly and as flexibly as possible. Each piece of this architecture must be resilient to server failures, load spikes, network partitions, and targeted attacks.

Cloud infrastructure helps address these needs. Its ability to provision capacity quickly in response to demand is an ideal match for the sudden and sometimes unexpected tidal waves of users that materialize on the web. In addition, cloud providers’ add-on services include a variety of convenient recipes that meet common requirements, greatly simplifying the design, deployment, and operation of web systems.
19.1 HTTP: the Hypertext Transfer Protocol

HTTP is the core network protocol for communication on the web. Lurking beneath a deceptively simple facade of stateless requests and responses lie layers of refinements that bring both flexibility and complexity. A well-rounded understanding of HTTP is a core competency for all system administrators.

In its simplest form, HTTP is a client/server, one-request/one-response protocol. Clients, also called user agents, submit requests for resources to an HTTP server. Servers receive incoming requests and process them by retrieving files from local disks, resubmitting them to other servers, querying databases, or performing any number of other possible computations. A typical page view on the web entails dozens or hundreds of such exchanges.

As with most Internet protocols, HTTP has adapted over time, albeit slowly. The centrality of the protocol to the modern Internet makes updates a high-stakes proposition. Official revisions are a slog of committee meetings, mailing list negotiations, public review periods, and maneuvering by stakeholders with vested and conflicting interests. During the long gaps between official revisions documented in RFCs, unofficial protocol extensions are born from necessity, become ubiquitous, and are eventually included as features in the next specification.

HTTP versions 1.0 and 1.1 are sent over the wire in plain text. Adventurous administrators can interact with servers directly by running `telnet` or `netcat`. They can also observe and collect HTTP exchanges by using protocol-agnostic packet capture software such as `tcpdump`.

See this page for general information about TLS.

The web is in the process of adopting HTTP/2, a major protocol revision that preserves compatibility with previous versions but introduces a variety of performance improvements. In an effort to promote the universal use of HTTPS (secure, encrypted HTTP) for the next generation of the web, major browsers such as Firefox and Chrome have elected to support HTTP/2 only over TLS-encrypted connections.

HTTP/2 moves from plain text to binary format in an effort to simplify parsing and improve network efficiency. HTTP’s semantics remain the same, but because the transmitted data is no longer directly legible to humans, generic tools such as `telnet` are no longer useful. The handy h2i command-line utility, part of the Go language networking repository at `github.com/golang/net`, helps restore some interactivity and debuggability to HTTP/2 connections. Many HTTP-specific tools such as `curl` also support HTTP/2 natively.
Uniform Resource Locators (URLs)

A URL is an identifier that specifies how and where to access a resource. URLs are not HTTP-specific; they are used for other protocols as well. For example, mobile operating systems use URLs to facilitate communication among apps.

You may sometimes see the acronyms URI (Uniform Resource Identifier) and URN (Uniform Resource Name) used as well. The exact distinctions and taxonomic relationships among URLs, URIs, and URNs are vague and unimportant. Stick with “URL.”

The general pattern for URLs is `scheme:address`, where `scheme` identifies the protocol or system being targeted and `address` is some string that’s meaningful within that scheme. For example, the URL mailto:ulsah@admin.com encapsulates an email address. If it’s invoked as a link target on the web, most browsers will bring up a preaddressed window for sending mail.

For the web, the relevant schemes are http and https. In the wild, you might also see the schemes ws (WebSockets), wss (WebSockets over TLS), ftp, ldap, and many others.

The address portion of a web URL allows quite a bit of interior structure. Here’s the overall pattern:

```html
scheme://[username:password@]hostname[:port][/path][?query][#anchor]
```

All the elements are optional except `scheme` and `hostname`.

See [this page](#) for more details about HTTP basic authentication.

The use of a `username` and `password` in the URL enables “HTTP basic authentication,” which is supported by most user agents and servers. In general, it’s a bad idea to embed passwords into URLs because URLs are apt to be logged, shared, bookmarked, visible in `ps` output, etc. User agents can get their credentials from a source other than the URL, and that is typically a better option. In a web browser, just leave the credentials out and let the browser prompt you for them separately.

HTTP basic authentication is not self-securing, which means that the password is accessible to anyone who listens in on the transaction. Therefore, basic authentication should really only be used over secure HTTPS connections.

The `hostname` can be a domain name or IP address as well as an actual hostname. The `port` is the TCP port number to connect to. The http and https schemes default to ports 80 and 443, respectively.

The `query` section can include multiple parameters separated by ampersands. Each parameter is a `key=value` pair. For example, Adobe InDesign users may find the following URL eerily familiar:
As with passwords, sensitive data should never appear as a URL query parameter because URL paths are often logged as plain text. The alternative is to transmit parameters as part of the request body. (You can’t really control this in other people’s web software, but you can make sure your own site behaves properly.)

The anchor component identifies a subtarget of a specific URL. For example, Wikipedia uses named anchors extensively as section headings, allowing specific parts of an entry to be linked to directly.
Structure of an HTTP transaction

HTTP requests and responses are similar in structure. After an initial line, both include a sequence of headers, a blank line, and finally, the body of the message, called the payload.

HTTP requests

The first line of a request specifies an action for the server to perform. It consists of a request method (also known as the verb), a path on which to perform the action, and the HTTP version to use. For example, a request to retrieve a top-level HTML page might look like this:

```
GET /index.html HTTP/1.1
```

Table 19.1 shows the common HTTP request methods. Verbs marked as “safe” should not change the server’s state. However, this is more a convention than a mandate. It’s ultimately up to the software that handles the request to decide how to interpret the verb.

Table 19.1: HTTP request methods

<table>
<thead>
<tr>
<th>Verb</th>
<th>Safe?</th>
<th>Purpose</th>
</tr>
</thead>
<tbody>
<tr>
<td>GET</td>
<td>Yes</td>
<td>Retrieves the specified resource</td>
</tr>
<tr>
<td>HEAD</td>
<td>Yes</td>
<td>Like GET, but requests no payload; retrieves metadata only</td>
</tr>
<tr>
<td>DELETE</td>
<td>No</td>
<td>Deletes the specified resource</td>
</tr>
<tr>
<td>POST</td>
<td>No</td>
<td>Applies request data to the given resource</td>
</tr>
<tr>
<td>PUT</td>
<td>No</td>
<td>Similar to POST, but implies replacement of existing contents</td>
</tr>
<tr>
<td>OPTIONS</td>
<td>Yes</td>
<td>Shows what methods the server supports for the specified path</td>
</tr>
</tbody>
</table>

GET is by far the most commonly used HTTP verb, followed by POST. REST APIs, discussed in Application programming interfaces (APIs), are more likely to employ the more exotic verbs such as PUT and DELETE.

The distinction between POST and PUT is subtle and largely of concern to web API developers. PUTs should be idempotent, meaning that a PUT can be repeated without causing ill effects. For example, a transaction that causes the server to send email should not be represented as a PUT. The rules for HTTP caching also differ significantly between PUT and POST. See RFC2616 for more details.

HTTP responses

The initial line in a response, called the status line, indicates the disposition of the request. It looks like this:

```
HTTP/1.1 200 OK
```
The important part is the three-digit numeric status code. The phrase that follows it is a helpful English translation that software ignores.

The first digit in the code determines its class; that is, the general nature of the result. Table 19.2 shows the five defined classes. Within a class, additional detail is provided by the remaining two digits. More than 60 status codes are defined, but only a few of these are commonly seen in the wild.

Table 19.2: HTTP response classes

<table>
<thead>
<tr>
<th>Code</th>
<th>General indication</th>
<th>Examples</th>
</tr>
</thead>
<tbody>
<tr>
<td>1xx</td>
<td>Request received; processing continues</td>
<td>101 Switching Protocols</td>
</tr>
<tr>
<td>2xx</td>
<td>Success</td>
<td>200 OK</td>
</tr>
<tr>
<td></td>
<td></td>
<td>201 Created</td>
</tr>
<tr>
<td>3xx</td>
<td>Further action needed</td>
<td>301 Moved Permanently</td>
</tr>
<tr>
<td></td>
<td></td>
<td>302 Found(^a)</td>
</tr>
<tr>
<td>4xx</td>
<td>Unsatisfiable request</td>
<td>403 Forbidden</td>
</tr>
<tr>
<td></td>
<td></td>
<td>404 Not Found</td>
</tr>
<tr>
<td>5xx</td>
<td>Server or environment failure</td>
<td>503 Service Unavailable</td>
</tr>
</tbody>
</table>

\(^a\) Most often used (inappropriately, according to the spec) for temporary redirects

**Headers and the message body**

Headers specify metadata about a request or response, such as whether to allow compression; what types of content are accepted, expected, or provided; and how intermediate caches should handle the data. For requests, the only required header is Host, which is used by web server software to determine which site is being contacted.

Table 19.3 shows some common headers.

Table 19.3: Commonly encountered HTTP headers
Table 19.3 is by no means a definitive list. In fact, both sides of the transaction can include any headers they wish. Both sides must ignore headers they don’t understand. By convention, custom and experimental headers were originally prefixed with “X-”. But some X- headers (such as X-Forwarded-For) became de facto standards, and it was then infeasible to remove the prefix because that would break compatibility. The use of X- is now deprecated by RFC6648.

Headers are separated from the message body by a blank line. For requests, the body can include parameters (for POST or PUT requests) or the contents of a file to upload. For responses, the message body is the payload of the resource being requested (e.g., HTML, image data, or query results). The message body is not necessarily human-readable, since it can contain images or other binary data. The body can also be empty, as for GET requests or most error responses.
curl: HTTP from the command line

**curl** (cURL) is a handy command-line HTTP client that’s available for most platforms. (Administrators will also encounter **libcurl**, a client library that developers can use to give their own software **curl**-like superpowers.) Here, we use **curl** to explore an HTTP exchange.

Below is an invocation of **curl** that requests the root of the web site admin.com on TCP port 80, which is the default for unencrypted (non-HTTPS) requests. The response payload (i.e., the admin.com homepage) and some informative messages from **curl** itself have been hidden by the **-o /dev/null** and **-s** flags. We also include the **-v** flag to request that **curl** display verbose output, which includes headers.

```
$ curl -s -v -o /dev/null http://admin.com
* Rebuilt URL to: http://admin.com/
* Hostname was NOT found in DNS cache
* Trying 54.84.253.153...
* Connected to admin.com (54.84.253.153) port 80 (#0)
> GET / HTTP/1.1
> User-Agent: curl/7.37.1
> Host: admin.com
> Accept: */*
>
< HTTP/1.1 200 OK
< Date: Mon, 27 Apr 2015 18:17:08 GMT
* Server Apache/2.4.7 (Ubuntu) is not blacklisted
< Server: Apache/2.4.7 (Ubuntu)
< Last-Modified: Sat, 02 Feb 2013 03:08:20 GMT
< ETag: "66d3-4d4b52c0c1100"
< Accept-Ranges: bytes
< Content-Length: 26323
< Vary: Accept-Encoding
< Content-Type: text/html
<
[ 2642 bytes data]
* Connection #0 to host admin.com left intact
```

Lines starting with > and < denote the request and response, respectively. In the request, the client tells the server that the user agent is **curl**, that it’s looking for host admin.com, and that it will accept any type of content as a response. The server identifies itself as Apache 2.4.7 and replies with contents of type HTML, along with a variety of other metadata.

We can set headers explicitly with **curl**’s **-H** argument. This feature is especially handy for making requests directly against IP addresses, bypassing DNS. For example, we could check that the server for www.admin.com responds identically to requests targeted at admin.com by setting the Host header, which informs the remote server of the domain the user agent is attempting to contact:
We use the `-O` argument to download a file. This example downloads a tarball of the `curl` source code to the current directory:

```
$ curl -H "Host: www.admin.com" -s -v -o /dev/null 54.84.253.153
```

<same output as the previous example, but with a different Host header>

We use the `-O` argument to download a file. This example downloads a tarball of the `curl` source code to the current directory:

```
$ curl -O http://curl.haxx.se/snapshots/curl-7.46.0-20151105.tar.gz
```

We’ve only scratched the surface of `curl`’s capabilities. It can handle other request methods such as POST and DELETE, store and submit cookies, download files, and assist with many different debugging scenarios.

Google’s Chrome browser offers a feature called “Copy as cURL” that creates a `curl` command to simulate the browser’s own behavior, including headers, cookies, and other details. You can easily retry requests with various adjustments and see the results exactly as the browser would. (Right-click a resource name in the Network tab of the developer tools panel to uncover this option.)
TCP connection reuse

TCP connections are expensive. In addition to the memory needed to maintain them, the three-way handshake used to establish each new connection adds latency equivalent to a full round trip before an HTTP request can even begin. (TCP Fast Open is a proposal that aims to improve this situation by allowing the SYN and SYN-ACK packets of TCP’s three-way handshake to also carry data. See RFC7413.)

The HTTP Archive, a project that tracks web statistics, estimates that the average site incurs requests for 99 resources per page load. If each resource required a new TCP connection, the performance of the web would be atrocious indeed. This was in fact the case early in the life of the web.

The original HTTP/1.0 specification did not include any provisions for connection reuse, but some adventurous developers added experimental support as an extension. The Connection: Keep-Alive header was added informally to clients and servers, then improved and made the default in HTTP/1.1. With keep-alive (also known as persistent) connections, HTTP clients and servers send multiple requests over a single connection, thus saving some of the cost and latency of initiating and tearing down multiple connections.

TCP overhead turns out to be nontrivial even when HTTP/1.1 persistent connections are enabled. Most browsers open as many as six parallel connections to the server to improve performance. Busy servers in turn must maintain many thousands of TCP connections in various states, resulting in network congestion and wasted resources.

HTTP/2 introduces multiplexing as a solution, allowing several transactions to be interleaved on a single connection. HTTP/2 servers can therefore support more clients per system, since each client imposes lower overhead.
HTTP over TLS

On its own, HTTP provides no network-level security. The URL, headers, and payload are open to inspection and modification at any point between the client and server. A malevolent infiltrator can intercept messages, alter their contents, or redirect requests to servers of its choice.

Enter Transport Layer Security (TLS), which runs as a separate layer between TCP and HTTP. TLS supplies only the security and encryption for the connection; it does not involve itself at the HTTP layer.

The precursor of TLS was known as SSL, the Secure Sockets Layer. All versions of SSL are obsolete and formally deprecated, but the name SSL remains in wide colloquial use. Outside of cryptographic contexts, assume that references to SSL really mean TLS.

The user agent verifies the server’s identity as part of the TLS connection process, eliminating the possibility of spoofing by counterfeit servers. Once the connection is established, its contents are protected against snooping and modification for the duration of the exchange. Attackers can still see the host and port used at the TCP layer, but they cannot access HTTP details such as the URL of a request or the headers that accompany it.

See Transport Layer Security for more details on TLS cryptography.
Virtual hosts

In the early days of the web, a server typically hosted only a single web site. When admin.com was requested, for example, clients performed a DNS lookup to find the IP address associated with that name and then sent an HTTP request to port 80 at that address. The server at that address knew that it was dedicated to hosting admin.com and served results accordingly.

As web use increased, administrators realized that they could achieve economies of scale if a single server could host more than one site at once. But how do you distinguish requests bound for admin.com from those bound for example.com if both kinds of traffic end up at the same network port?

One possibility is to define virtual network interfaces, effectively permitting several different IP addresses to be bound to a single physical connection. Most systems allow this, and it works fine, but the scheme is fiddly and requires management at several different layers.

A better solution, virtual hosts, was provided by HTTP 1.1 in RFC2616. This scheme defines a Host HTTP header that user agents set explicitly to indicate what site they're attempting to contact. Servers examine the header and behave accordingly. This convention conserves IP addresses and simplifies management, especially for sites that have hundreds or thousands of web sites on a single server.

HTTP 1.1 requires user agents to provide a Host header, so virtual hosts are now the standard way that web servers and administrators handle server consolidation.

The use of name-based virtual hosts in combination with TLS is a bit tricky under the hood. TLS certificates are issued to specific hostnames, which are chosen when the certificate is generated. A TLS connection must be established before the server can read the Host header from the HTTP request, but without that header, the server does not know which virtual host it should be impersonating, and hence, which certificate to select.

The solution is SNI, Server Name Indication, in which the client submits the hostname that it’s requesting as part of the initial TLS connection message. Modern servers and clients all handle SNI automatically.
19.2 Web software basics

A rich library of open source software facilitates the construction of flexible, resilient web applications. Table 19.4 lists a few general categories of services that speak HTTP and perform specific functions within the web application stack.

<table>
<thead>
<tr>
<th>Type</th>
<th>Purpose</th>
<th>Examples</th>
</tr>
</thead>
<tbody>
<tr>
<td>Application server</td>
<td>Runs web app code, interfaces to web servers</td>
<td>Unicorn, Tomcat</td>
</tr>
<tr>
<td>Cache</td>
<td>Speeds access to frequently requested content</td>
<td>Varnish, Squid</td>
</tr>
<tr>
<td>Load balancer</td>
<td>Relays requests to downstream systems</td>
<td>Pound, HAProxy</td>
</tr>
<tr>
<td>Web app firewall a</td>
<td>Inspects HTTP traffic for common attacks</td>
<td>ModSecurity</td>
</tr>
<tr>
<td>Web server</td>
<td>Serves static content, couples to other servers</td>
<td>Apache, NGiNX</td>
</tr>
</tbody>
</table>

| a. Often abbreviated WAF |

A web proxy is an intermediary that receives HTTP requests from clients, optionally performs some processing, and relays the requests to their ultimate destination. Proxies are normally transparent to clients. Load balancers, web application firewalls, and cache servers are all specialized types of proxy servers. A web server also acts as a sort of proxy if it relays requests to application servers.

Exhibit A illustrates the role that each service plays in an HTTP exchange. Requests can be fulfilled higher in the stack if the requested resource can be satisfied, or rejected with a 4xx or 5xx code if a problem occurs. Requests that require a query to the database traverse every layer.

Exhibit A: Components of a web application stack
To maximize availability, each layer should run on more than one node simultaneously. Ideally, redundancy should span geographical regions so that the overall design is not dependent on any single physical data center. This goal is a lot easier to achieve when you build on a cloud platform that offers well-defined geographic regions as a fundamental building block (as most do).

Real-world architectures aren’t usually as straightforward as Exhibit A suggests. In addition, most web software components perform functions in more than one area. NGINX is best known as a web server, for example, but it’s also a highly capable cache and load balancer. An NGINX web server with caching features enabled is more efficient than a stack of separate servers running on individual virtual machines.
Web servers and HTTP proxy software
Most sites use web servers either to proxy HTTP connections to application servers or to serve static content directly. A few of the features provided by web servers include

- Virtual hosts, allowing many sites to coexist peacefully within a single server
- Handling of TLS connections
- Configurable logging that tracks requests and responses
- HTTP basic authentication
- Routing to different downstream systems according to requested URLs
- Execution of dynamic content through application servers

The leading open source web servers are the Apache HTTP Server, known colloquially as httpd, and NGINX, which is pronounced “engine X.”

Netcraft, an English Internet research and security company, publishes monthly market share statistics for web servers. As of June 2017, Netcraft shows that around 46% of active web sites run Apache. NGINX accounts for 20% and has been steadily rising since 2008.

Apache httpd is the original project from the Apache Software Foundation, now known for supporting a variety of excellent open source projects. httpd has been under active development since 1995 and is widely regarded as the reference HTTP server implementation.

NGINX is a versatile server designed for speed and efficiency. Like httpd, NGINX supports service of static web content, load balancing, monitoring of downstream servers, proxying, caching, and other related functions.

Some development systems, notably Node.js and the language Go, implement web servers internally and can handle many HTTP workflows without the need for a separate web server. These systems incorporate sophisticated connection management features and are robust enough for production workloads.

The H2O server (h2o.examp1e.net; note the numeral one in “examp1e”) is a newer web server project that takes full advantage of HTTP/2 features and achieves even better performance than NGINX. Because it was first released in 2014, it can’t claim the track record of Apache or NGINX. On the other hand, neither is it constrained by historical implementation decisions as is httpd. It’s certainly worth a look for new deployments.

It’s hard to make strong recommendations among these options because they’re all quite good. That said, for mainstream production use, we suggest NGINX. It offers exceptional performance and a relatively simple and modern configuration system.
Load balancers

You can’t run a highly available web site on a single server. Not only does this configuration expose your users to every potential hiccup experienced by the server, but it also gives you no way to update the software, operating system, or configuration without downtime.

Single servers are also exquisitely vulnerable to load spikes and intentional attacks. The more overloaded a server becomes, the more time it spends thrashing instead of getting useful work done. Past a certain load threshold (one that you will have to discover through bitter experience!), performance craters abruptly rather than degrading gracefully.

To avoid these problems, you can use a load balancer, which is a type of proxy server that distributes incoming requests among a set of downstream web servers. Load balancers also monitor the status of those servers to ensure that they are providing timely and correct responses.

Exhibit B shows the placement of load balancers in an architecture diagram.

### Exhibit B: The role of a load balancer

<table>
<thead>
<tr>
<th>Internet</th>
<th>DMZ</th>
<th>Secure net</th>
</tr>
</thead>
<tbody>
<tr>
<td></td>
<td></td>
<td></td>
</tr>
</tbody>
</table>

Load balancers solve many of the problems inherent in a single-system design:

- Load balancers do not process requests but merely route them to other systems. As a result, they can handle many more concurrent requests than does a typical web server.

- When a web server needs a software upgrade or has to be taken off-line for any other reason, it can easily be removed from the rotation.

- If one of the servers experiences a problem, the health-check mechanism on the load balancer detects the problem and removes the errant system from the server pool until it becomes healthy again.
To avoid becoming single points of failure themselves, load balancers usually run in pairs. Depending on the configuration, one balancer might act as a passive backup while the other serves live traffic, or both balancers might serve requests simultaneously.

The way that requests are distributed is usually configurable. Here are a few common algorithms:

- **Round robin**, in which requests are distributed among the active servers in a fixed rotation order
- **Load equalization**, in which new requests go to the downstream server that’s currently handling the smallest number of connections or requests
- **Partitioning**, in which the load balancer selects a server according to a hash of the client’s IP address. This method ensures that requests from the same client always reach the same web server.

Load balancers normally operate at layer four of the OSI model, in which they route requests based just on the IP address and port of the request. However, they can also operate at layer seven by inspecting requests and routing them according to their target URL, cookie values, or other HTTP headers. For example, requests for example.com/linux might route to a separate set of servers than do requests for example.com/bsd.

See [this page](#) for more information about network DMZs (demilitarized zones).

As an added bonus, load balancers can improve security. They usually reside in the DMZ portion of a network and proxy requests to web servers behind an internal firewall. If HTTPS is in use, they also perform TLS termination: the connection from the client to the load balancer uses TLS, but the connection from the load balancer to the web server can be vanilla HTTP. This arrangement offloads some processing overhead from the web servers.

Load balancers can distribute other kinds of traffic in addition to (or instead of) HTTP. A common use is to add a load balancer that distributes requests to databases such as MySQL or Redis.

The most common open source load balancers for UNIX and Linux are NGINX, already introduced as a web server, and HAProxy, a high-performance TCP and HTTP proxy beloved by veteran administrators for its flexible configuration, stability, and robust performance. Both are excellent and well documented, and both have large user communities. (Apache **httpd** also has a load-balancing module, though we haven’t seen it used as widely.)

Commercial load balancers such as those from F5 and Citrix are available both as hardware devices to be installed in a data center and as software solutions. They typically offer a graphical configuration interface, more features than open source tools, extra functions in addition to straightforward load balancing, and hefty price tags.
Amazon offers a dedicated load-balancing service, the Elastic Load Balancer (ELB), for use with EC2 virtual machines. ELB is a completely managed service; no virtual machine is required for the load balancer itself. ELB handles an extremely large number of concurrent connections and can balance traffic among multiple availability zones.

In ELB terminology, a “listener” accepts connections from clients and proxies them to back-end EC2 instances that do the actual work. Listeners can proxy TCP, HTTP, or HTTPS traffic. The load is distributed according to the “least connections” algorithm.

ELB is not the most fully featured load balancer, but it is our recommended solution for AWS-hosted systems because it requires virtually no administrative attention.
Caches

Web caches were born from the observation that clients often repeatedly access the same content within a short time. Caches live between clients and web servers and store the results of the most frequent requests, sometimes in memory. They can then intervene to answer requests for which they know the correct response, reducing load on the authoritative web servers and improving response times for users.

In caching jargon, an origin is the original content provider, the source of truth about the content. Caches get their content directly from the origin or from another upstream cache.

Several factors determine caching behavior:

- The values of HTTP headers, including Cache-Control, ETag, and Expires
- Whether the request is served by HTTPS, for which caching is more nuanced
- The response status code; some are not cacheable (see RFC2616)
- The contents of HTML `<meta>` tags (not respected by all caches)

Static blobs such as images, videos, CSS stylesheets, and JavaScript files are well suited to caching because they rarely change. Dynamic content loaded from a database or another system in real time is more difficult—but not necessarily impossible—to cache.

Because HTTPS payloads are encrypted, responses cannot be cached unless the cache server terminates the TLS connection, decrypting the payload. The connection from the cache server to the origin may then require a separately encrypted TLS connection (or not, depending on whether the connection between the two is trusted).

For highly variable pages that should never be cached, developers set the following HTTP header:

```
Cache-Control: no-cache, no-store
```

Exhibit C shows the placement of several potential cache layers in an HTTP request.

---

### Exhibit C: Caching players involved in handling an HTTP request

<table>
<thead>
<tr>
<th>Local network</th>
<th>Internet</th>
<th>Server network</th>
</tr>
</thead>
<tbody>
<tr>
<td>Browser cache</td>
<td>Proxy cache</td>
<td>Reverse proxy cache</td>
</tr>
<tr>
<td>Origin server</td>
<td></td>
<td></td>
</tr>
</tbody>
</table>
**Browser caches**

All modern web browsers save recently used resources (images, stylesheets, JavaScript files, and some HTML pages) locally to speed up backtracking and return visits.

In theory, browser caches should follow exactly the same caching rules as does any other HTTP cache. This is why your browser’s Back button usually evinces a slight lag instead of zipping you instantly to the previous page. Even though most of the resources needed to render the page are cached locally, the page’s top-level HTML wrapper is typically dynamic and uncacheable, so a round trip to the server is still required. The browser could simply rerender the materials on hand from the previous visit—and one or two used to do that—but this shortcut breaks the correctness of caching and leads to a variety of subtle problems.

**Proxy cache**

You can install a proxy cache at the edge of an organization’s network to speed up access for all users. When a user loads a web site, the requests are first received by the proxy cache. If a requested resource is cached, that resource is immediately returned to the user without the remote site being consulted.

You can configure a proxy cache in two ways: actively, by changing users’ browser settings to point to the proxy; or passively, by having a network router send all web traffic through the cache server. The latter configuration is known as an intercepting proxy. There are also methods by which user agents can automatically discover the relevant proxies.

**Reverse proxy cache**

Web site operators configure a “reverse proxy” cache to offload traffic from their web and application servers. Incoming requests are first routed to the reverse proxy cache, from which they may be served immediately if the requested resources are available. The reverse proxy passes requests for uncached resources along to the appropriate web server.

Server sites use reverse proxy caches primarily because they reduce load on the origin servers. They may also have the beneficial side effect of speeding response times for clients.

**Cache problems**

Web caches are tremendously important to the performance of the web, but they also introduce complexity. A problem at any caching layer can introduce stale content that is out of date with respect to the origin server. Cache problems can befuddle both users and administrators and are sometimes difficult to debug.

Stale cache entries are best detected by a direct query at each hop along the path. If you are the site operator, try using `curl` to request a problematic page directly from the origin, then from the reverse proxy cache, and if applicable, from the proxy cache and from any other caches in the request path.
You can use `curl -H "Cache-Control: no-cache"` to politely request a cache refresh. This is the same as invoking `<Shift-Reload>` in a web browser. Conformant caches will obey, but if you’re still seeing old data, don’t assume that your reload request has been honored unless you can prove it on the server.

**Cache software**

Table 19.5 lists a few of the open source caching software implementations. Of these, we find ourselves using NGINX most frequently. Its caching is easy to configure, and NGINX is often already in use as a proxy or web server.

<table>
<thead>
<tr>
<th>Server</th>
<th>Notes</th>
</tr>
</thead>
<tbody>
<tr>
<td>Squid</td>
<td>One of the first open source cache implementations</td>
</tr>
<tr>
<td></td>
<td>Normally used as a proxy cache</td>
</tr>
<tr>
<td></td>
<td>Includes important features like antivirus and TLS</td>
</tr>
<tr>
<td>Varnish</td>
<td>Exceptional configuration language</td>
</tr>
<tr>
<td></td>
<td>Multithreaded</td>
</tr>
<tr>
<td></td>
<td>Modular and extensible</td>
</tr>
<tr>
<td>Apache mod_cache</td>
<td>Good choice for sites already running <strong>httpd</strong></td>
</tr>
<tr>
<td>NGINX</td>
<td>Good choice for sites already running NGINX</td>
</tr>
<tr>
<td></td>
<td>Has a reputation for good performance</td>
</tr>
<tr>
<td>Apache Traffic Server</td>
<td>Runs at extremely high-traffic sites</td>
</tr>
<tr>
<td></td>
<td>Supports HTTP/2</td>
</tr>
<tr>
<td></td>
<td>Donated to the Apache Foundation by Yahoo!</td>
</tr>
</tbody>
</table>
Content delivery networks

A content delivery network (CDN) is a globally distributed system that improves web performance by moving content closer to users. Nodes in a CDN are dispersed geographically to hundreds or thousands of locations. When clients request content from a site that uses a CDN, they are routed to the closest node (called an edge server), thereby decreasing latency and reducing congestion for the origin.

Edge servers are similar to proxy caches. They store copies of content locally. If they don’t have a local copy of a requested resource or if their version of the content has expired, they retrieve the resource from the origin, respond to the client, and update their cache.

CDNs use DNS to redirect clients to the geographically nearest host. Exhibit D explains how it works.

Exhibit D: The role of DNS in a content delivery network

CDNs can now host dynamic content, but traditionally they have been best suited to static content such as images, stylesheets, JavaScript libraries, HTML files, and downloadable objects. Streaming services like Netflix and YouTube use CDNs to serve large media files.

CDNs also offer value beyond performance improvements. Most CDNs provide security services such as denial-of-service attack prevention and web application firewalls. Some specialty CDNs offer other innovations that optimize page rendering and reduce the load on origin servers.

A substantial portion of content on the web today is served by CDNs. If you’re at a large site, expect to pull out your pocketbook to pay for the privilege of fast performance. If you run a smaller service, optimize your local caching layers before turning to a CDN.

One of the oldest and most prestigious (read: expensive) CDNs is Akamai, headquartered in Massachusetts. Akamai counts some of the world’s largest governments and businesses among its customers. It has the largest global network as well as some of the most advanced CDN features. Nobody was ever fired for choosing Akamai.
CloudFlare is another popular CDN. Unlike Akamai, CloudFlare has a history of selling to smaller customers, although their target market has more recently shifted to enterprise. Pricing is listed clearly on their web site, and they offer some of the best security features available. CloudFlare was one of the first large-scale providers to deploy HTTP/2 for all its customers.

Amazon’s CDN service is called CloudFront. It integrates with other AWS services such as S3, EC2, and ELB, but can also work for sites hosted outside of Amazon’s cloud. As with other AWS products, pricing is competitive and metered by usage.
Languages of the web

The web has evolved from being mostly static to a rich, interactive experience. The web apps that enable this bounty are coded in a variety of programming languages, each with associated tooling and unique quirks. Administrators need to manage software libraries, install application servers, and configure web applications according to the standards established for each language’s ecosystem.

All the languages mentioned in the following sections are in common use on the web today. They all feature engaged communities of developers, extensive support libraries, and well-documented best practices. Sites typically choose whichever languages and frameworks their teams are most comfortable with.

Ruby

See the section starting here for a short summary of Ruby.

Ruby is well known in DevOps and system administration circles for its use in Chef and Puppet. It’s also the language of Ruby on Rails, a widely used web framework. Rails is a good choice for rapid development processes and is often used for prototyping new ideas.

Rails has a reputation for mediocre performance and for fostering monolithic applications. Over time, many Rails applications tend to accumulate baggage that makes them harder to modify; the end result is often a gradual performance sag over time.

Ruby features a large collection of libraries called gems that developers can use to simplify their projects. Most are hosted at rubygems.org. They are curated by the community, but many are of marginal quality. Managing a system’s installed versions of Ruby and its various gem dependencies can be both tedious and troublesome.

Python

See the section starting here for a short summary of Python.

Python is a general-purpose language used not only in web development but also in a wide swath of scientific disciplines. It’s easy to read and to learn. The most widely deployed web framework for Python is Django, which has many of the same benefits and drawbacks as Ruby on Rails.

Java

Java, now controlled by Oracle, is used most often in enterprise environments with slower development workflows. Java offers fast performance at the expense of complex, clunky tooling
and many layers of abstraction. Java’s challenging license requirements and obtuse conventions can frustrate neophytes.

**Node.js**

JavaScript is known first and foremost as a client-side scripting language that runs within web browsers. As a language, it has been ridiculed as hastily designed, difficult to read, and frequently counterintuitive. Now, Node.js—an engine for executing JavaScript on servers—brings JavaScript to the data center as well.

To be fair, Node.js boasts high concurrency and native real-time messaging capabilities. As a newer language, it has so far avoided much of the cruft built up over the years in other systems.

**PHP**

PHP is simple to get started with, and for that reason it tends to attract new and inexperienced programmers. PHP applications are notorious for being difficult to maintain. Past versions of PHP made it far too easy for developers to create large security holes in their applications, but recent versions have made improvements in this area. PHP is the language used by WordPress, Drupal, and several other content management systems.

**Go**

Go is a lower-level language from Google. It has gained popularity in recent years through its use in major open source projects such as Docker. It’s excellent for systems programming but is also well suited for web applications because of its powerful native concurrency primitives. One benefit for administrators is that Go software usually compiles to a stand-alone binary, making it simple to deploy.
**Application programming interfaces (APIs)**

Web APIs are application interfaces intended for use not by humans but by software agents. An API defines a set of methods through which a remote system can make use of an application’s data and services. APIs have become ubiquitous on the web because they promote cooperation among many diverse clients.

APIs are nothing new. Operating systems define APIs to allow user-space applications to interact with the kernel. Nearly all software packages use defined interfaces to facilitate modularity and separation of functions within the code base. However, web APIs are a bit special because they are exposed to the world on the public web with the intention of promoting use by outside developers.

Web API calls are normal HTTP requests. They’re only “APIs” because the client and server have agreed, by convention, that certain URLs and verbs have specific meanings and effects within the domain of their interaction.

Web APIs commonly use some kind of text-based serialization format to encode data for exchange. These formats are relatively simple and can be parsed by applications written in any programming language. Many formats exist, but by far the most common are JavaScript Object Notation (JSON) and Extensible Markup Language (XML). (Douglas Crockford, who named and promoted the JSON format, says it’s pronounced like the name Jason. But somehow, “JAY-sawn” seems to have become more common in the technical community.)

HTTP APIs are perhaps easiest to explain by example. The Spotify music service exposes an API that represents its music library. A client of the API can request information about albums, artists, and tracks; execute searches; and perform other related actions. This API is used both by Spotify’s own client applications (its browser, desktop, and mobile clients) and by third parties that want to incorporate Spotify’s services.

Because web APIs consist of HTTP requests, you can interact with them with all the normal HTTP tools, including web browsers and **curl**. For example, we can obtain Spotify’s JSON record for The Beatles:
Here, we’ve piped the JSON output through `jq` to clean up the formatting a bit. On a terminal, `jq` also colorizes the output. `jq` does far more than just formatting and is highly recommended for parsing and filtering JSON from the command line. Find it at [stedolan.github.io/jq](https://stedolan.github.io/jq).

How did we know that 3WrFJ7ztbogyGnTHbHJFl2 is the Spotify ID for The Beatles? Try searching through the API: [https://api.spotify.com/v1/search?type=artist&q=beatles](https://api.spotify.com/v1/search?type=artist&q=beatles).

Spotify’s API is also an example of a “RESTful” API, which is the predominant approach today. REST (Representational State Transfer) is an architectural style of API design introduced by Roy Fielding in his doctoral dissertation; Fielding is also a primary author of the HTTP specification. The term was originally quite specific but is now more loosely applied to web services that 1) explicitly use HTTP verbs to communicate intent, and 2) use a directory-like path structure to locate resources. Most REST APIs use JSON as their underlying representation for data.

REST contrasts starkly with SOAP (Simple Object Access Protocol), an earlier system for implementing HTTP APIs that defines strict and elaborate multilevel guidelines for interactions among systems. SOAP APIs use a complex XML-based format that funnels all calls through a few specific URLs, resulting in large HTTP payloads, poor performance, and endless difficulties in development, debugging, and deployment.

The development of the SOAP ecosystem is an interesting case study of the ways that technical initiatives can go awry. In particular, it illustrates the risks of attempting to design systems for a hazy and uncertain future. SOAP put a lot of effort into remaining platform-, language-, data-, and transport-neutral, and indeed it largely achieved these goals—even basic data types such as integers were left open to definition. Unfortunately, the resulting system was complex and didn’t fit well with real-world needs.
19.3 Web hosting in the cloud

Cloud providers offer dozens of services for hosting web applications, and the landscape changes weekly. We can’t possibly cover everything, but a few points stand out as being of particular interest to web administrators.

Small sites with few users and a tolerance for occasional outages can get away with a single virtual cloud instance as a web server (or possibly two instances behind a load balancer for improved reliability). But the cloud offers many opportunities to improve these simple configurations without significant increases in cost and complexity of administration.
Build versus buy

Administrators working on a cloud platform can build custom, self-managed web applications out of “raw” virtual machines. Alternatively, they can farm out parts of the design to off-the-shelf cloud services, thus reducing the labor involved in designing, configuring, and maintaining everything by hand. For the sake of efficiency, we prefer to rely on vendor services when possible.

Load balancers are a good example of this tradeoff. On AWS, for example, you can either run an EC2 instance with open source load-balancing software or sign up for an AWS-provided Elastic Load Balancer. The former offers greater customization but requires you to manage the load balancer’s operating system, configure the load-balancing software, tune performance, and promptly install security patches as they are released in the future. In addition, the glue needed to gracefully handle failures of the software or the instance will be somewhat more complex.

An ELB, on the other hand, can be created in a matter of seconds and requires no further administrative action. AWS handles everything behind the scenes. Unless the ELB lacks a specific feature that you need, it is clearly the expedient choice.

Ultimately, this is a decision between building a service or outsourcing it to the vendor. For the sake of your own sanity, avoid the building option unless the function in question is a core competence for your business.
Platform-as-a-Service

The PaaS concept simplifies web hosting for developers by eliminating infrastructure as a concern. Developers package their code in a specific format and upload it to the PaaS provider, which provisions appropriate systems and runs it automatically. The provider issues a DNS endpoint connected to the client’s running application, which the client can then customize by using a DNS CNAME record.

Although PaaS offerings greatly simplify infrastructure management, they sacrifice flexibility and customization. Most offerings either do not allow administrative access to a shell or they actively discourage it. Users of a PaaS must accept certain design decisions made by the vendor. Users’ ability to implement some features may be constrained.

Google App Engine pioneered the PaaS concept and remains one of the most prominent products. App Engine supports Python, Java, PHP, and Go, and has many supporting features such as cron-like scheduled job execution, programmatic access to logs, real-time messaging, and access to various databases. It is considered the Cadillac of PaaS offerings.

The competing product from AWS is called Elastic Beanstalk. In addition to all the languages supported by App Engine, it supports Ruby, Node.js, Microsoft .NET, and Docker containers. It integrates with Elastic Load Balancers and AWS’s Auto Scaling feature, leveraging the power of the AWS ecosystem.

In practice, we’ve found Elastic Beanstalk to be a mixed bag. Customization is possible through an extension framework that is proprietary and tedious. Users are still responsible for running the EC2 instances that host the application. So although Elastic Beanstalk might be a fine fit for prototyping, we believe that the system is not a good choice for production workloads consisting of many services.

Heroku is another respected vendor in this space. An application on Heroku is deployed to a dyno, Heroku’s word for a lightweight Linux container. Users control the dyno deployments. Heroku has a strong network of partnerships that offer databases, load balancing, and other integrations. Heroku’s pricing is higher than some other offerings in part because its own infrastructure runs on AWS under the hood.
Static content hosting

See this page for more information about content delivery networks.

It seems like overkill to run an operating system just for the sake of hosting static web sites. Fortunately, the cloud providers can host them for you. In AWS S3, you create a bucket for your content, then configure a CNAME record from your domain to an endpoint within the provider. In Google Firebase, you use a command-line tool to copy your local content to Google, which provisions an SSL certificate and hosts your files. In both cases you can serve your content from a CDN for better performance.
Serverless web applications

AWS Lambda is an event-based computing service. Developers who use Lambda write code that runs in response to an event such as the arrival of a message in a queue, a new object in a bucket, or even an HTTP request. Lambda feeds the event payload and metadata as inputs to a user-defined function, which performs processing and returns a response. There are no instances or operating systems to manage.

To process HTTP requests, Lambda is used in conjunction with another AWS service called API Gateway, a proxy that can scale to hundreds of thousands of simultaneous requests. API Gateway is interposed in front of an origin to add features such as access control, rate limiting, and caching. HTTP requests are received by the API Gateway, and when a request arrives, the gateway triggers a Lambda function.

In combination with static hosting on S3, Lambda and API Gateway can lead to a fully serverless platform for running web applications, as illustrated in Exhibit E.

Exhibit E: Serverless web hosting with AWS Lambda, API Gateway, and S3

This technology is still in its youth, but it’s already altering the mechanics of hosting web applications. We expect enhancements, frameworks, competing services, and best practices to mature rapidly in the coming years.
19.4 **Apache HTTPD**

The **httpd** web server is ubiquitous among the many flavors of UNIX and Linux. It is portable across many architectures, and prebuilt packages exist for all major systems. Unfortunately, OS vendors have varied and highly opinionated approaches to **httpd** configuration.

A modular architecture has been fundamental to Apache’s adoption. Dynamic modules can be turned on through configuration, offering alternative authentication options, improved security, support for running code written in most languages, URL-rewriting superpowers, and many other features.

For largely historical reasons, Apache has a pluggable connection handling system called multi-processing modules (MPMs) that determines how HTTP connections are managed at the network I/O layer. The event MPM is the modern choice and is recommended over the worker and prefork alternatives. (Some legacy software that is not considered thread-safe, such as mod_php, should use the prefork MPM. It uses processes rather than threads for each connection.)

To bind to privileged ports (those below 1024, such as HTTP port 80 and HTTPS port 443), the initial **httpd** process must run as root. That process then forks additional workers under a local account with lower privileges to handle actual requests. Sites that do not need to listen on port 80 or 443 can be run entirely without root privileges.

**httpd** is configured through directives (Apache-speak for configuration options) in plain text files that use a distinctive Apache-style syntax. Though hundreds of directives exist, administrators usually need to tweak only a few. The directives and their values are documented directly in the default files that ship with the OS as well as on Apache’s web site.
httpd in use

httpd is both the name given to the daemon’s binary and to the project. Ubuntu has taken the liberty of renaming httpd to apache2, which matches the name of the apt package but otherwise does little more than create confusion.

System V init, BSD init, and systemd can all manage httpd. Whichever option is standard for your system is the one you should default to. For debugging and configuration, however, you can interact with the daemon independently of the startup scripts.

Administrators can either run httpd directly or use apachectl. Invoking httpd offers direct control over the server daemon, but remembering (and typing!) all the options is a challenge. apachectl is a shell script wrapper around httpd. Each operating system vendor customizes apachectl to conform to the conventions of its init process. It can start, stop, reload, and show the status of Apache.

For example, here’s how to start the server with the default configuration:

```
# apachectl start
Performing sanity check on apache24 configuration:
Syntax OK
Starting apache24.
# apachectl status
apache24 is running as pid 1337.
```

In this output from a FreeBSD system, apachectl first performs a lint-like configuration check by running httpd -t, then starts the daemon. (lint is a UNIX program that evaluates C code for potential bugs. The term is now applied more broadly to any tool that inspects software and configuration files for errors, bugs, or other problems.)

apachectl graceful waits for any currently open connections to conclude and then restarts the server. This feature is handy for updating without interrupting active connections. It’s available through the system start and stop scripts as well.

Use apachectl’s -f flag to start Apache with a custom configuration, e.g.:

```
# apachectl -f /etc/httpd/conf/custom-config.conf -k start
```

Some vendors deprecate this use of apachectl in favor of running httpd directly.

Refer to Chapter 2, Booting and System Management Daemons, to learn how to configure httpd to start automatically at boot time.
httpd configuration logistics

Although an entire httpd configuration can be contained in a single file, OS maintainers typically use the include directive to split the default configuration into multiple files and directories. This architecture simplifies site management and is better suited to automation. Predictably, the specifics of the configuration hierarchy differ by system. Table 19.6 lists the Apache configuration defaults for each of our example platforms.

Table 19.6: Apache configuration details by platform

<table>
<thead>
<tr>
<th></th>
<th>RHEL/CentOS</th>
<th>Debian/Ubuntu</th>
<th>FreeBSD</th>
</tr>
</thead>
<tbody>
<tr>
<td>Package name</td>
<td>httpd</td>
<td>apache2</td>
<td>apache24</td>
</tr>
<tr>
<td>Config root</td>
<td>/etc/httpd</td>
<td>/etc/apache2</td>
<td>/usr/local/etc/apache24</td>
</tr>
<tr>
<td>Primary config file</td>
<td>conf/httpd.conf</td>
<td>apache2.conf</td>
<td>httpd.conf</td>
</tr>
<tr>
<td>Module config</td>
<td>conf.modules.d/</td>
<td>mods-available/mods-enabled/</td>
<td>modules.d/</td>
</tr>
<tr>
<td>Virtual host config</td>
<td>conf.d/</td>
<td>sites-available/sites-enabled/</td>
<td>Includes/</td>
</tr>
<tr>
<td>Log location</td>
<td>/var/log/httpd</td>
<td>/var/log/apache2</td>
<td>/var/log/httpd-* .log</td>
</tr>
<tr>
<td>User</td>
<td>apache</td>
<td>www-data</td>
<td>www</td>
</tr>
</tbody>
</table>

When httpd starts, it consults a primary configuration file, usually httpd.conf, and incorporates any additional files as referenced by include directives. The default httpd.conf is heavily commented and serves as a quick reference. Configuration options in this file can be grouped into three categories:

- Global settings such as the path to httpd’s configuration root, the user and group as which to run, the modules to activate, and the network interfaces and ports to listen on
- VirtualHost sections that define how to provide service for a given domain (usually delegated to subdirectories and included in the main configuration)
- Instructions for answering requests that don’t match any VirtualHost definition

Many admins will be satisfied with the global settings and need only manage individual VirtualHosts.

Modules exist independently of the httpd core and often have their own configuration options. Most OS vendors choose to separate out module configuration into subdirectories.

Debian and Ubuntu approach Apache configuration idiosyncratically. A structure of subdirectories, configuration files, and symlinks creates a more flexible system for managing the
server, at least in theory.

Exhibit F attempts to clarify this puzzle. The master `apache2.conf` file includes all files from the `*-enabled` subdirectories in `/etc/apache2`. These files are in fact symbolic links to files in the `*-available` subdirectories. A pair of configuration commands that create and remove symlinks is provided for each set of subdirectories.

Exhibit F: Subdirectories of `/etc/apache2` on Debian-based systems

In our experience, the Debian system is unnecessary and overly complex. A simple `site-configuration` subdirectory usually provides sufficient structure. If you’re running Debian or Ubuntu, though, it makes sense to stick with their defaults.
Virtual host configuration

The lion’s share of httpd configuration lies in virtual host definitions. It’s generally a good idea to create a file for each site.

When an HTTP request arrives, httpd identifies which virtual host to select by consulting the HTTP Host header and network port. It then matches the path portion of the requested URL to a Files, Directory, or Location directive to determine how to serve the requested content. This mapping process is known as request routing.

The following sample shows the HTTP and HTTPS configuration for admin.com.

```xml
<VirtualHost *:80>
    ServerName admin.com
    ServerAlias www.admin.com
    ServerAlias ulsah.admin.com
    Redirect / https://admin.com/
</VirtualHost>

<VirtualHost *:443>
    ServerName admin.com
    ServerAlias www.admin.com
    ServerAlias ulsah.admin.com
    DocumentRoot /var/www/admin.com/
    CustomLog /var/log/apache2/admin_com_access combined
    ErrorLog /var/log/apache2/admin_com_error
    SSLEngine on
    SSLCertificateFile "/etc/ssl/certs/admin.com.crt"
    SSLCertificateKeyFile "/etc/ssl/private/admin.com.key"
    <Directory "/var/www/admin.com/"
        Require all granted
    </Directory>
    <Directory "/var/www/admin.com/photos/"
        Options +Indexes
    </Directory>
    <IfModule mod_rewrite.c>
        RewriteEngine on
        RewriteRule ^/(usah|lsah) /ulsah
    </IfModule>
    ExtendedStatus On
    <Location /server-status>
        SetHandler server-status
        Require ip 10.0.10.10/32
    </Location>
</VirtualHost>
```

Much of this is self explanatory, but a few details are worth noting:
The first VirtualHost answers on port 80 and redirects all HTTP requests for admin.com, www.admin.com, and ulsah.admin.com to use HTTPS.

Requests for admin.com/photos receive an index of all files in that directory.

Requests for /usah or /lsah are rewritten to /ulsah.

Server status, accessible in this configuration at www.admin.com/server-status, is a module that shows useful runtime performance information, including statistics about the daemon’s CPU and memory usage, request status, the average number of requests per second, and more. Monitoring systems can use this feature to collect data about the web server for alerting, reporting, and visualization of HTTP traffic. Here, access to server status is restricted to a single IP address, 10.0.10.10.

**HTTP basic authentication**

In the HTTP basic authentication scheme, clients pass a base-64-encoded username and password in the Authorization HTTP header. If a user includes a name and password in a URL (e.g., https://user:pass@www.admin.com/server-status), the browser performs the encoding and transfers the value to the Authorization header automatically.

The username and password are not encrypted, so basic authentication does not provide any confidentiality. Thus, it is safe to use only in combination with HTTPS.

Basic authentication in Apache is configured in `Location` or `Directory` blocks. For example, the following snippet requires authentication to access /server-status (a best practice) and limits access to a subnet:

```xml
<Location /server-status>
    SetHandler server-status
    Require ip 10.0.10.0/24
    AuthType Basic
    AuthName "Restricted"
    AuthUserFile /var/www/.htpasswd
    Require valid-user
</Location>
```

Note that the account information is stored externally to the configuration file. Use `htpasswd` to create the account entries:
Password files are conventionally hidden files called `.htpasswd`, but they can be named anything. Even though the passwords are encrypted, set the permissions on `.htpasswd` files to be readable only by the web-server user. This precaution limits attackers’ ability to see usernames and to run passwords through cracking software.

**Configuring TLS**

SSL might have changed its name to TLS, but in the interest of backward compatibility, Apache retains the SSL name for its configuration options (as do many other software packages). Just a few lines are needed to set up TLS:

- `SSLEngine on`
- `SSLProtocol all -SSLv2 -SSLv3`
- `SSLCertificateFile "/etc/ssl/certs/admin.com.crt"
- `SSLCertificateKeyFile "/etc/ssl/private/admin.com.key"

Here, the TLS certificate and key are located in Linux’s central system location, `/etc/ssl`. The public certificates can be readable by anyone, but the key should be accessible only to the Apache master-process user, typically root. We prefer to set permissions to 444 for the certificate and 400 for the key.

All versions of the actual SSL protocol (precursor to TLS) are known to be insecure and should be disabled with the `SSLProtocol` directive, shown above.

See [this page](#) for a complete citation for the Server Side TLS guide.

A few ciphers have known weaknesses. You can configure the web server’s supported ciphers with the `SSLCipherSuite` directive. The best practices for precisely which settings to use are constantly in flux. The Mozilla Server Side TLS guide is the best resource that we are aware of for staying current on best practices for TLS. It also has a handy configuration syntax reference for Apache, NGINX, and HAProxy.

**Running web applications within Apache**

httpd can be extended to run programs written in Python, Ruby, Perl, PHP, and other languages from within the module system. Modules run inside Apache processes and have access to the full
HTTP request/response life cycle.

Modules provide additional configuration directives that let administrators control the runtime characteristics of applications. Table 19.7 lists some common application server modules.

**Table 19.7: Application server modules for httpd**

<table>
<thead>
<tr>
<th>Module</th>
<th>Lang</th>
<th>Description</th>
</tr>
</thead>
<tbody>
<tr>
<td>mod_php</td>
<td>PHP</td>
<td>Deprecated; use only with the prefork MPM</td>
</tr>
<tr>
<td>mod_wsgi</td>
<td>Python</td>
<td>The Web Server Gateway Interface, a Python standard</td>
</tr>
<tr>
<td>mod_passenger</td>
<td>Multiple</td>
<td>Flexible, commercially supported application server for multiple languages, including Ruby, Python, and Node.js</td>
</tr>
<tr>
<td>mod_proxy_fcgi</td>
<td>Any</td>
<td>A standard server interface usable from any language</td>
</tr>
<tr>
<td>mod_perl</td>
<td>Perl</td>
<td>A Perl interpreter that lives within httpd</td>
</tr>
</tbody>
</table>

The following example (to configure a Python Django application for api.admin.com) uses mod_wsgi:

```
LoadModule wsgi_module mod_wsgi.so

<VirtualHost *:443>
  ServerName api.admin.com

  CustomLog /var/log/apache2/api_admin_com_access combined
  ErrorLog /var/log/apache2/api_admin_com_error

  SSLEngine on
  SSLCertificateFile "/etc/ssl/certs/api_admin_com.crt"
  SSLCertificateKeyFile "/etc/ssl/private/api_admin_com.key"

  WSGIDaemonProcess admin_api user=user1 group=group1 threads=5
  WSGIScriptAlias / /var/www/api.admin.com/admin_api.wsgi

  <Directory /var/www/api.admin.com>
    WSGIProcessGroup admin_api
    WSGIApplicationGroup %{GLOBAL}
    Require all granted
  </Directory>
</VirtualHost>
```

Once mod_wsgi.so has been loaded by Apache, several WSGI configuration directives become available. The WSGIScriptAlias file in the configuration above, admin_api.wsgi, contains Python code that is needed by the WSGI module.
Logging

httpd offers best-in-class logging capabilities, with fine-grained control over the data that is logged and the ability to separate log data by virtual host. Administrators use these logs to debug configuration problems, detect potential security threats, and analyze usage information.

A sample log message from admin.com.access.log looks like this:

```
127.0.0.1 - - [19/Jun/2015:15:21:06 +0000] "GET /search HTTP/1.1"
   200 20892 "-" "curl/7.38.0"
```

The message shows:

- The source of the request; in this case, 127.0.0.1, the local host
- A time stamp
- The path of the requested resource (/search) and the HTTP method (GET)
- The response status code (200)
- The size of the response
- The user agent (the curl command-line tool)

The documentation for mod_log_config has all the details on how to customize the log format.

A busy web site generates a large number of request logs that can quickly fill up the disk. Administrators are responsible for ensuring that this never happens. Keep web server logs on a dedicated partition to prevent a large log file from affecting the rest of the system.

See this page for more information about logrotate.

On most Linux distributions, the default package installation of Apache includes an appropriate logrotate configuration. FreeBSD comes with no such default, and administrators should instead add an entry in /etc/newsyslog.conf for Apache’s logs.

The log directory and the files within should be writable only by the user of the master httpd process, which is normally root. If nonroot users have write access, they can create a symlink to another file, causing it to be overwritten with bogus data. The system defaults are set safely, so avoid customizing the owner and group.
19.5 NGINX

A busy web server must respond to many thousands of concurrent requests. Most of the time needed to handle each request is spent waiting for data to arrive from the network or disk. The time spent actively processing the request is short by comparison.

To handle this workload efficiently, NGINX uses an event-based system in which just a few worker processes handle many requests simultaneously. When a request or response (an event) is ready for servicing, a worker process quickly completes processing before returning to handle the next event. Above all, NGINX aims to avoid blocking on network or disk I/O.

The event MPM included in newer releases of Apache uses a similar architecture, but for high-volume and performance-sensitive sites, NGINX remains the software of choice.

Administrators running NGINX will notice at least two processes: a master and a worker. The master performs housekeeping duties such as opening sockets, reading the configuration, and keeping the other NGINX processes running. Workers do most of the heavy lifting by handling and processing requests. Some configurations use additional processes dedicated to caching. As in Apache, the master process runs as root so that it can open sockets for any ports below 1024. The other processes run as a less privileged user.

The number of worker processes is configurable. A good rule of thumb is to run as many worker processes as the system has CPU cores. Debian and Ubuntu configure NGINX this way by default if it’s installed from a package. FreeBSD and RHEL default to a single worker process.
Installing and running NGINX

Although NGINX continues to grow in popularity and is a staple among some of the world’s busiest web sites, OS distributions still lag on NGINX support. The versions available in the official repositories for Debian and RHEL are usually out of date, though FreeBSD is typically more current. NGINX is open source, so it can be built and installed manually. The project’s web page, nginx.org, offers packages for **apt** and **yum** than are generally more current than those supplied by the distributions.

See [this page](#) for more information about signals.

The system’s normal service management is appropriate for day-to-day wrangling of **nginx**. You can also run the **nginx** daemon during development and debugging. Use the `-c` argument to specify a custom configuration file. The `-t` option performs a check of the configuration file syntax.

**nginx** uses signals to trigger various maintenance actions; [Table 19.8](#) lists these. Make sure you target the master **nginx** process (usually the one with the lowest PID).

<table>
<thead>
<tr>
<th>Signal</th>
<th>Function</th>
</tr>
</thead>
<tbody>
<tr>
<td>TERM or INT</td>
<td>Shuts down immediately</td>
</tr>
<tr>
<td>QUIT</td>
<td>Completes and closes all current connections, then shuts down</td>
</tr>
<tr>
<td>USR1</td>
<td>Reopens log files (used to facilitate log rotations)</td>
</tr>
<tr>
<td>HUP</td>
<td>Reloads the configuration(^a)</td>
</tr>
<tr>
<td>USR2</td>
<td>Gracefully replaces the server binary without interrupting service(^b)</td>
</tr>
</tbody>
</table>

\(^a\) This option tests the syntax of the new configuration, and if the syntax is valid, starts new workers with the new configuration. It then gracefully shuts down the old workers.

\(^b\) See the **nginx** command-line documentation for details on how this works.
Configuring NGINX

The NGINX configuration style is generally C-like; it uses curly braces to distinguish blocks of configuration lines and semicolons to separate lines. The main configuration file is called `nginx.conf` by default. Table 19.9 summarizes the most important system-specific aspects of NGINX configuration.

### Table 19.9: NGINX configuration details by platform

<table>
<thead>
<tr>
<th>Package name</th>
<th>RHEL/CentOS</th>
<th>Debian/Ubuntu</th>
<th>FreeBSD</th>
</tr>
</thead>
<tbody>
<tr>
<td>Daemon path</td>
<td>/sbin/nginx</td>
<td>/usr/sbin/nginx</td>
<td>/usr/local/sbin/nginx</td>
</tr>
<tr>
<td>Configuration root</td>
<td>/etc/nginx</td>
<td>/etc/nginx</td>
<td>/usr/local/etc/nginx</td>
</tr>
<tr>
<td>Virtual host config</td>
<td>conf.d/</td>
<td>sites-available/</td>
<td>No prescribed location</td>
</tr>
<tr>
<td></td>
<td></td>
<td>sites-enabled/</td>
<td></td>
</tr>
<tr>
<td>Default user</td>
<td>nginx</td>
<td>www-data</td>
<td>nobody</td>
</tr>
</tbody>
</table>

a. You must enable the EPEL software repository; see fedoraproject.org/wiki/EPEL.
b. Relative to the configuration root directory

Within the `nginx.conf` file, blocks of configuration directives surrounded by curly braces are called contexts. A context contains directives specific to that block of configuration. For example, here’s a minimal (but complete) NGINX configuration that shows three contexts:

```nginx
events {}

http {
    server {
        server_name www.admin.com;
        root /var/www/admin.com;
    }
}
```

The outermost context (called `main`) is implicit and configures the core functionality. The `events` and `http` contexts live within `main`. `events` is a required context that configures connection handling. Since it’s blank in this example, default values are implied. Fortunately, the defaults are sensible:

- Run one worker process (use the unprivileged user account).
- Listen on port 80 if started as root or port 8000 otherwise.
- Write logs to `/var/log/nginx` (chosen at compile time).

The `http` context contains all directives relating to web and HTTP proxy services. `server` contexts, which define virtual hosts, are nested within `http`. Multiple `server` contexts within `http` would configure multiple virtual hosts.
Aliases can be included in `server_name` to match the Host header against a group of subdomains:

```nginx
http {
    server {
        server_name admin.com www.admin.com;
        root /var/www/admin.com;
    }
    server {
        server_name example.com www.example.com;
        root /var/www/example.com;
    }
}
```

See this page for an overview of regular expressions.

The value for `server_name` can also be a regular expression, and the match can even be captured and named as a variable for use later in configuration. By using this feature, you can refactor the previous configuration to

```nginx
http {
    server {
        server_name ^(www\.)?(?<domain>(example|admin).com)\$;
        root /var/www/$domain;
    }
}
```

The regular expression, which starts with a tilde, matches either example.com or admin.com, optionally preceded by www. The value of the matched domain is stored in the `$domain` variable, which is then used to determine which server root to select.

Be aware that use of this syntax commits NGINX to performing a regular expression match on every HTTP request. We use it here to demonstrate NGINX’s flexibility, but in practice you would probably want to just list all possible hostnames in plain text. It’s perfectly reasonable to use regular expressions in `nginx.conf`, but make sure they’re delivering actual value, and try to keep them low in the configuration hierarchy so that they activate only in specific situations.

Name-based virtual hosts can be distinguished from IP-based hosts by using the `listen` and `server_name` directives together.
This configuration shows two versions of admin.com being served from different web roots. The IP address of the interface on which the request was received determines which version of the site the client sees.

The root is the base directory where HTML, images, stylesheets, scripts, and other files for the virtual host are stored. By default, NGINX just serves files out of the root, but you can use the location directive to do more sophisticated request routing. If a given path isn’t matched by a location directive, NGINX automatically falls back to the root.

The following example uses location in combination with the proxy_pass directive. It instructs NGINX to serve most requests from the web root but forward requests for http://www.admin.com/nginx to nginx.org.

```
server {
    server_name admin.com www.admin.com;
    root /var/www/admin.com/site1;
}
server {
    listen 10.0.10.11:80
    server_name admin.com www.admin.com;
    root /var/www/admin.com/site2;

    location /nginx/ {
        proxy_pass http://nginx.org/;
    }
}
```

proxy_pass instructs NGINX to act as a proxy and replay requests from clients to another downstream server. We revisit the proxy_pass directive when we describe how to use NGINX as a load balancer here.

location can use regular expressions to perform powerful path-based routing to different sources based on the requested content. The official NGINX documentation analyzes how NGINX evaluates the server_name, listen, and location directives to route requests.

A common pattern among distributions is to set sensible defaults for many directives in the global http context, then use the include directive to add site-specific virtual hosts to the final configuration. For example, the default nginx.conf file for Ubuntu includes the line

```
include /etc/nginx/conf.d/*.conf;
```
This architecture helps eliminate redundancy since all children inherit the global settings. Administrators in straightforward environments may not need to do anything more than write virtual host configurations expressed as server contexts.
Configuring TLS for NGINX

Although NGINX didn’t borrow much from Apache’s configuration style, its TLS configuration is one area in which it’s strikingly similar. As in Apache, the configuration keywords all refer to SSL, TLS’s earlier name.

Enable TLS and point to the certificate and private key file like so:

```nginx
server {
    listen 443;
    ssl on;
    ssl_certificate /etc/ssl/certs/admin.com.crt;
    ssl_certificate_key /etc/ssl/private/admin.com.crt;
    ssl_protocols TLSv1 TLSv1.1 TLSv1.2;
ssl_ciphers ECDHE-RSA-AES128-GCM-SHA256:ECDHE... # truncated
    ssl_prefer_server_ciphers on;

    server_name admin.com www.admin.com;
    root /var/www/admin.com/site1;
}
```

Only the actual TLS protocols (not the older SSL versions) should be enabled; all SSL protocols have been deprecated. Permissions on the certificate and key should follow the recommendations outlined in the Apache TLS section here.

Use the `ssl_ciphers` directive to require cryptographically strong cipher suites and to disable weaker ciphers. The `ssl_prefer_server_ciphers` option in conjunction with `ssl_ciphers` instructs NGINX to choose from the server’s list rather than from the client’s; otherwise, the client could suggest any cipher it pleased. (The previous example does not show a full list of ciphers because the appropriate list is quite long; refer to the Mozilla Server Side TLS guide cited here for recommended values. If you prefer a shorter cipher list, try the one at cipherli.st.)
**Load balancing with NGINX**

In addition to being a web and cache server, NGINX is also a capable load balancer. Its configuration style is flexible but somewhat nonobvious.

Use the `upstream` module to create named groups of servers. For example, the following clause defines `admin-servers` as a collection of two servers:

```
upstream admin-servers {
    server web1.admin.com:8080 max_fails=2;
    server web2.admin.com:8080 max_fails=2;
}
```

`upstream` groups can be referenced from virtual host definitions. In particular, they can be used as proxying destinations, just like hostnames:

```
http {
    server {
        server_name admin.com www.admin.com;
        location / {
            proxy_pass http://admin-servers;
            health_check interval=30 fails=3 passes=1 uri=/health_check
            match=admin-health  # line not split in original file
        }
    }
    match admin-health {
        status 200;
        header Content-Type = text/html;
        body ~ "Red Leader, Standing By";
    }
}
```

Here, traffic for admin.com and www.admin.com is farmed out to the web1 and web2 servers in round robin order (the default).

This configuration also sets up health checks for the back-end servers. Checks are performed every 30 seconds (`interval=30`) against each server at the `/health_check` endpoint (`uri=/health_check`). NGINX will mark the server down if the health check fails on three consecutive attempts (`fails=3`), but will add the server back to the rotation if it succeeds just once (`passes=1`).

The `match` keyword is peculiar to NGINX. It dictates the conditions under which the health check is considered successful. In this case, NGINX must receive a 200 response code, the Content-Type header must be set to text/html, and the body of the response must contain the phrase “Red Leader, Standing By.”

We’ve added an additional condition within the `upstream` context that sets the maximum number of connection attempt failure to two. That is, if NGINX cannot connect to the server at all within
two attempts, it gives up and removes that server from the pool. This is an additive connectivity check that complements the more structured checks from the `health_check` clause.
19.6 HAProxy

HAProxy is the most widely used open source load-balancing software. It proxies HTTP and TCP, supports sticky sessions to pin a given client to a specific web server, and offers advanced health-checking capabilities. Recent versions also support TLS, IPv6, and HTTP compression. Support for HTTP/2 is a work in progress and is expected to mature quickly beginning with HAProxy version 1.7.

HAProxy’s configuration is usually contained in a single file, `haproxy.cfg`. It’s so simple that OS vendors generally don’t overcomplicate things and instead embrace the default directory structure recommended by the project.

On Debian and RHEL systems, the configuration is in `/etc/haproxy/haproxy.cfg`. FreeBSD doesn’t provide a default, as there really is no sensible one for load balancing; it’s entirely dependent on your setup. You can find an example configuration on FreeBSD in `/usr/local/share/examples/haproxy` after the HAProxy package has been installed.

The following simple example configuration sets HAProxy to listen on port 80 and distribute requests in a round robin fashion between two web servers, web1 and web2, on port 8080.

```
global
  daemon
  maxconn 5000
defaults
  mode http
  timeout connect 5000 # Milliseconds
  timeout client 10000
  timeout server 10000
frontend http-in
  bind *:80
  default_backend webservers
backend webservers
  balance roundrobin
  server web1 10.0.0.10:8080
  server web2 10.0.0.11:8080
```

This example introduces HAProxy’s `frontend` and `backend` keywords, illustrated in Exhibit G.

Exhibit G: HAProxy frontend and backend specifications
frontend dictates how HAProxy will receive requests from clients: which addresses and ports to use, what types of traffic to serve, and other client-facing considerations. backend configures the set of servers that actually process requests. Multiple frontend/backend pairs can exist in a single configuration, allowing a single HAProxy to service multiple sites.

The timeout settings allow fine-grained control over how long a system should wait when trying to open a new connection to a server and how long to keep connections open once they have been established. Fine-tuning these values is important on busy web servers. On local networks, the timeout connect value can be quite low (500ms or less) because new connections should be established quickly.
Health checks

Although the previous configuration provides basic functionality, it doesn’t check the status of downstream web servers. If web1 or web2 goes off-line, half of incoming requests would begin to fail.

HAProxy’s status-check feature performs regular HTTP requests to determine the health of each server. As long as servers respond with an HTTP 200 response code, they remain in service and continue to receive requests from the load balancer.

If a server fails a status check (by returning anything other than status 200), then HAProxy removes the errant server from the pool. However, HAProxy continues to perform health checks on the server. If it starts to respond successfully once again, HAProxy will return it to the pool.

The specifics of the health check, such as what request method to use, the interval between checks, and the path to request, can all be adjusted. In this example, HAProxy performs a GET request for / on each server every 30 seconds:

```
backend webservers
    balance roundrobin
    option httpchk GET /
    server web1 10.0.0.10:8080 check inter 30000
    server web2 10.0.0.11:8080 check inter 30000
```

It’s reassuring to know that you can contact a machine’s web server, but that’s hardly the last word on server health. Well-constructed web applications commonly expose a health-check endpoint that performs a thorough probe of the application to determine its true health. These checks may include verification of database or cache connectivity as well as performance monitoring. Use these more sophisticated checks if they are available.
Server statistics

HAProxy offers a convenient web interface that displays server stats, much like mod_status in Apache. HAProxy’s version shows the state of each server in the pool and lets you manually enable and disable servers as needed.

The syntax is straightforward:

```
listen stats :8000
  mode http
  stats enable
  stats hide-version
  stats realm HAProxy\ Statistics
  stats uri /
  stats auth myuser:mypass
  stats admin if TRUE
```

Server stats can be configured either within a specific listener or within a backend or frontend block, to limit the feature to that configuration alone.
**Sticky sessions**

HTTP is a stateless protocol, so each transaction is an independent session. From the perspective of the protocol, requests from the same client are unrelated.

At the same time, most web applications need state to track user behavior over time. The classic example of state is a shopping cart. Users browse a store, add items to the cart, and when ready to check out, submit their payment information. The web application needs some way to track the contents of the cart across multiple page views.

Most web applications use cookies to track state. The web application generates a session for a user and puts the session ID in a cookie that is sent back to the user in the response header. Each time a client submits a request to the server, the cookie is sent with the request. The server uses the cookie to recover the client’s context.

Ideally, web applications should store their state information in a persistent and shared medium such as a database. However, some poorly behaved web applications keep their session data locally, in the server’s memory or on its local disk. When placed behind a load balancer, these applications break because a single client’s requests might be routed to multiple servers, depending on the vagaries of the load balancer’s scheduling algorithm.

To address this issue, HAPerxy can insert a cookie of its own into responses, a feature known as sticky sessions. Any future requests from the same client will include the cookie. HAPerxy can use the value of the cookie to route the request back to the same server.

A version of the previous configuration modified to support sticky sessions looks like the following. Note the addition of the `cookie` directive.

```
backend webservers
    balance roundrobin
    option httpchk GET /
    cookie SERVERNAME insert httponly secure
    server web1 10.0.0.10:8080 cookie web1 check inter 30000
    server web2 10.0.0.11:8080 cookie web2 check inter 30000
```

In this configuration, HAPerxy maintains a SERVERNAME cookie to track the server that a client is dealing with. The `secure` keyword specifies that the cookie should only be sent over TLS connections, and `httponly` informs browsers to use the cookie only over HTTP. Refer to RFC6265 for further information on these attributes.
TLS termination

HAProxy versions 1.5 and later include TLS support. A common configuration is to terminate TLS connections at the HAProxy server and communicate with back-end servers over plain HTTP. This approach offloads the cryptographic overhead from the back-end servers and reduces the number of systems that need a private key.

For particularly security-conscious sites, it’s also possible to use HTTPS from HAProxy to the back-end servers. You can use the same TLS certificate or a different one; either way, you will still need to terminate and reinitiate TLS at the proxy.

Since HAProxy terminates the TLS connection from clients, you’ll need to add the pertinent configuration to the `frontend` configuration block.

```
frontend https-in
    bind *:443 ssl crt /etc/ssl/private/admin.com.pem
    default_backend webservers
```

Apache and NGINX require the private key and certificate to be in separate files in PEM format, but HAProxy expects both components to be present in the same file. You can simply concatenate the separate files to create a composite file:

```
# cat /etc/ssl/{private/admin.com.key,certs/admin.com.crt} > /etc/ssl/private/admin.com.pem
# chmod 400 /etc/ssl/private/admin.com.pem
# ls -l /etc/ssl/private/admin.com.pem
-r-------- 1 root root 3660 Jun 18 17:46 /etc/ssl/private/admin.com.pem
```

Since the private key is part of the composite file, ensure that the file is owned by root and is not readable by any other user. (If you do not run HAProxy as root because you are not accessing any privileged ports, make sure the ownership of the key file matches the identity under which HAProxy runs.)

All usual best practices for TLS apply to HAProxy: disable SSL-era protocols and explicitly configure the acceptable cipher suites.
19.7 Recommended Reading

Adrian, David, et al. Weak Diffie-Hellman and the Logjam Attack. weakdh.org. This page describes the Logjam attack on the Diffie-Hellman key exchange protocol and suggests ways to secure systems properly.

CloudFlare, Inc. blog.cloudflare.com. This is the corporate blog of content delivery network CloudFlare. Some posts are just marketing information, but many include insights on the latest web trends and technologies.

Google, Inc. Web Fundamentals. developers.google.com/web/fundamentals. This is a useful guide to various best practices for web development, including sections on site design, user interfaces, security, performance, and other topics of interest to both developers and administrators. The caching discussion is particularly good.


IANA. Index of HTTP Status Codes. www.iana.org/assignments/http-status-codes.


Stenberg, Daniel. daniel.haxx.se/blog. This is the blog of Daniel Stenberg, the author of curl and a prolific HTTP expert.

Van Elst, Remy. Strong Ciphers for Apache, nginx, and Lighthttpd. cipherli.st. Correct and secure cipher configuration for Apache httpd, NGINX, and the lighthttpd web servers, as well as a TLS configuration tester.
SECTION THREE
STORAGE
Data storage systems are looking more and more like a giant set of Lego blocks that you can assemble in an infinite variety of configurations. You can build anything from a lightning-fast storage space for a mission-critical database to a vast, archival vault that stores three copies of all data and can be rewound to any point in the past.

Mechanical hard drives remain a popular storage medium when capacity is the most important consideration, but solid state drives (SSDs) are preferred for performance-sensitive applications. Caching systems, both software and hardware, help combine the best features these storage types.

On cloud servers, you usually have a choice of storage hardware, but you’ll pay more for SSD-backed virtual disks. You can also choose from a variety of purpose-specific storage types, such as object stores, infinitely expandable network drives, and relational databases-as-a-service.

Running on top of this real and virtual hardware are a variety of software components that mediate between the raw storage devices and the filesystem hierarchy seen by users. These components include device drivers, partitioning conventions, RAID implementations, logical volume managers, systems for virtualizing disks over a network, and the filesystem implementations themselves.

In this chapter, we discuss the administrative tasks and decisions that occur at each of these layers. We begin with “fast path” instructions for adding a basic disk to Linux or FreeBSD. We
then review storage-related hardware technologies and look at the general architecture of storage software. We then work our way up the storage stack from low-level formatting to the filesystem level. Along the way, we cover disk partitioning, RAID systems, and logical volume managers.

Above the level of individual machines lie a variety of schemes for sharing data on a network. Chapters 21 and 22 describe two common file sharing systems: NFS for native sharing among UNIX and Linux systems, and SMB for interoperability with Windows and macOS systems.
20.1 I just want to add a disk!

Before we launch into many pages of storage architecture and theory, we first address the most common scenario: you want to install a hard disk and make it accessible through the filesystem. Nothing fancy: no RAID, all the drive’s space in a single volume, and the default filesystem type.

Step one is to attach the drive. If the machine in question is a cloud server, you generally provision a virtual drive of the desired size within the provider’s administrative GUI (or through their API) and then attach it to an existing virtual server as a separate step. It’s normally unnecessary to reboot the server because cloud (and virtual) kernels recognize such hardware changes on the fly.

In the case of physical hardware, drives that communicate through a USB port can simply be powered on and plugged in. SATA and SAS drives need to be mounted in a bay, enclosure, or cradle. Although some hardware and drivers are designed to permit hot-addition of SATA drives, that feature requires hardware support and is uncommon in mass-market hardware. Reboot the system to make sure the OS is in a configuration that’s reasonably reproducible at boot time.

If you’re running a desktop machine with a window system and all the stars align, the system might offer to format a new disk for you when you plug it in. That’s particularly likely if you’re plugging in an external USB disk or thumb drive. The autoformat option usually works fine; use it if it’s offered. However, check the mount details afterwards (by running the `mount` command in a terminal window) to make sure the drive hasn’t been mounted with restrictions you don’t want (e.g., with execution or normal ownerships disabled).

If you set up the disk by hand, it’s critically important to identify and format the right disk device. A newly added drive is not necessarily represented by the highest-numbered device file, and on some systems, the addition of a new drive can change the device names of existing drives (after a reboot, usually). Double-check the identity of the new drive by reviewing its manufacturer, size, and model number before you do anything that’s potentially destructive! Use the commands mentioned in the next two sections.
Linux recipe

First, run `lsblk` to list the system’s disks and identify the new drive. If that output doesn’t give you enough information to conclusively identify the new drive, you can list model and serial numbers with `lsblk -o +MODEL,SERIAL`.

See this page for an explanation of GPT partition tables.

Once you know which device file refers to the new disk (assume it’s `/dev/sdb`), install a partition table on the disk. Several commands and utilities can do this, including `parted`, `gparted`, `fdisk`, `cfdisk`, and `sfdisk`; it doesn’t matter which one you use, as long as it understands GPT-style partition tables. `gparted` is probably the easiest option on a system with a graphical user interface. Below, we show the `fdisk` recipe, which works on all Linux systems. (Some systems still ship a version of `parted` that doesn’t understand GPT.)

```bash
$ sudo fdisk /dev/sdb
Welcome to fdisk (util-linux 2.23.2).
Changes will remain in memory only, until you decide to write them.
Be careful before using the write command.
Command (m for help): g
Building a new GPT disklabel (GUID:

AB780438-DA90-42AD-8538-ECC9626228C7)
Command (m for help): n
Partition number (1-128, default 1): <Return>
First sector (2048-1048575966, default 2048): <Return>
Last sector, +sectors or +size[K,M,G,T,P] (2048-1048575966, default 1048575966): <Return>
Created partition 1
Command (m for help): w
The partition table has been altered!
Calling ioctl() to re-read partition table.
Syncing disks.
```

The `g` subcommand creates a GPT partition table. The `n` subcommand creates a new partition; pressing `<Return>` in response to all `fdisk`’s questions allocates all free space to the new partition (partition 1). Finally, the `w` subcommand writes the new partition table to the disk.

The device file for the newly created partition is the same as the device file for the disk as a whole with a 1 appended to it. In the example above, the partition is `/dev/sdb1`.

You can now create a filesystem on `/dev/sdb1` with the `mkfs` command. The `-L` option gives the filesystem a shorthand label (here, “spare”). The label stays the same even if the disk that contains the filesystem is assigned a different device name during a subsequent boot.
Next, create a mount point and mount the new filesystem:

```
$ sudo mkdir /spare
$ sudo mount LABEL=spare /spare
```

You could equivalently specify /dev/sdb1 instead of **LABEL=spare** as a way of identifying the partition, but that name won’t necessarily work in the future.

To have the filesystem automatically mounted at boot time, edit the /etc/fstab file and duplicate one of the existing entries. Change the device name and mount point to match those shown in the `mount` command above. For example,

```
LABEL=spare   /spare ext4    errors=remount-ro  0  0
```

You can also use a UUID to identify the filesystem; see [this page](#).

See [this page](#) for more details on Linux device files for disks. See [this page](#) for partitioning information. See [this page](#) for information about the ext4 filesystem.
FreeBSD recipe

Run `geom disk list` to list the disk devices that the kernel is aware of. Unfortunately, FreeBSD doesn’t divulge much information beyond device names and sizes. You can resolve any ambiguity as to which disk is which by running `geom part list` to see which devices have existing partitions. An unformatted disk should have no partitions.

Once you know the disk name, you can install a partition table and create a filesystem. In this example, we assume that the disk name is `ada1` and that you want to mount the new filesystem as `/spare`.

```bash
$ sudo gpart create -s GPT ada1  # Create GPT partition table
ada1 created

$ sudo gpart add -l spare -t freebsd-ufs -a 1M ada1  # Create partition
ada1p1 added

$ sudo newfs -L spare /dev/ada1p1  # Create filesystem
/dev/ada1p1: 5120.0MB (10485680 sectors) block size 32768, fragment size 4096
  using 9 cylinder groups of 626.09MB, 20035 blks, 80256 inodes.
  super-block backups (for fsck_ffs -b #) at:
  192, 1282432, 2564672, 3846912, 5129152, 6411392, 7693632,
  8975872, 10258112
...
```

The `-l` option to `gpart add` applies a text label to the new partition. The label makes the partition accessible through the path `/dev/gpt/spare` regardless of what device name the kernel assigns to the underlying disk device. The `-L` option to `newfs` applies a similar (but distinct) label to the new filesystem to make the partition accessible as `/dev/ufs/spare`.

Mount the filesystems with the following commands:

```bash
$ sudo mkdir /spare
$ sudo mount /dev/ufs/spare /spare
```

To have the filesystem automatically mounted at boot time, add it to the `/etc/fstab` file (see this page).
20.2 Storage Hardware

Even in today’s post-Internet world, computer data can be stored in only a few basic ways: hard disks, flash memory, magnetic tapes, and optical media. The last two technologies have significant limitations that disqualify them from use as a system’s primary filesystem. However, they’re still sometimes used for backups and for “near line” storage—cases in which instant access and rewritability are not of primary concern.

After 40 years of traditional magnetic disk technology, performance-minded system builders finally received a practical alternative in the form of solid state disks (SSDs). These flash-memory-based devices offer a different set of tradeoffs from those of a standard disk, and they will be influencing the architectures of databases, filesystems, and operating systems for years to come.

At the same time, traditional hard disks are continuing their exponential increases in capacity. Thirty years ago, at the dawn of the 5.25” form factor that remains in use today, a 60MB hard disk cost $1,000. Today, a garden-variety 4TB drive runs $125 or so. That’s more than 500,000 times more storage for the money, or double the TB/$ every 1.6 years. During that same period, the sequential throughput of mass-market drives has increased from 500 kB/s to 200 MB/s, a comparatively paltry factor of 400. And random-access seek times have hardly budged. The more things change, the more they stay the same.

See this page for more information on IEC units (gibibytes, etc.).

Disk sizes are specified in gigabytes that are billions of bytes, as opposed to memory, which is specified in “gigabytes” (gibibytes, really) of $2^{30} \times (1,073,741,824)$ bytes. The difference is about 7%. Be sure to check your units when estimating and comparing capacities.

Hard disks and SSDs are enough alike that they can act as drop-in replacements for each other, at least at the hardware level. They use the same hardware interfaces and interface protocols. And yet they have very different strengths, as Table 20.1 summarizes.

Table 20.1: Comparison of HDD and SSD technology
In the next sections, we take a closer look at each of these technologies along with a more recent category of storage devices: hybrid drives.

<table>
<thead>
<tr>
<th>Characteristic</th>
<th>HDD</th>
<th>SSD</th>
</tr>
</thead>
<tbody>
<tr>
<td>Typical size</td>
<td>&lt; 16TB</td>
<td>&lt; 2TB</td>
</tr>
<tr>
<td>Random access time a</td>
<td>8ms</td>
<td>0.25ms</td>
</tr>
<tr>
<td>Sequential read</td>
<td>200 MB/s</td>
<td>450 MB/s</td>
</tr>
<tr>
<td>Random read</td>
<td>2 MB/s</td>
<td>450 MB/s</td>
</tr>
<tr>
<td>IOPS b</td>
<td>150 ops/s</td>
<td>100,000 ops/s</td>
</tr>
<tr>
<td>Cost</td>
<td>$0.03/GB</td>
<td>$0.26/GB</td>
</tr>
<tr>
<td>Reliability</td>
<td>Poor</td>
<td>Poor c</td>
</tr>
<tr>
<td>Limited writes</td>
<td>No</td>
<td>In theory</td>
</tr>
</tbody>
</table>

a. Performance and cost values are as of mid 2017
b. I/O operations per second
c. Fewer whole-device failures than HDD, but more data loss
**Hard disks**

A typical hard drive contains several rotating platters coated with magnetic film. They are read and written by tiny skating heads mounted on a metal arm that swings back and forth to position them. The heads float close to the surface of the platters but don’t actually touch them.

Reading from a platter is quick; it’s the mechanical maneuvering needed to address a particular sector that drives down random-access throughput. Delays come from two main sources.

First, the head armature must swing into position over the appropriate track. This part is called seek delay. Second, the system must wait for the right sector to pass underneath the head as the platter rotates. That part is rotational latency. Disks can stream data at hundreds of MB/s if reads are optimally sequenced, but random reads are unlikely to achieve more than a few MB/s.

A set of tracks on different platters that are all the same distance from the spindle is called a cylinder. The cylinder’s data can be read without any additional movement of the arm. Although heads move amazingly fast, they still move much more slowly than the disks spin around. Therefore, any disk access that does not require the heads to seek to a new position will be faster.

Spindle speeds vary. 7,200 RPM remains the mass-market standard for enterprise and performance-oriented drives. A few 10,000 RPM and 15,000 RPM drives remain available at the high end, but the advent of inexpensive SSDs now limits these drives to a small and shrinking niche market. Higher rotational speeds decrease latency and increase the bandwidth of data transfers, but the drives tend to run hot.

**Hard disk reliability**

Hard disks fail frequently. A 2007 Google Labs study of 100,000 drives surprised the tech world with the news that hard disks more than two years old had an average annual failure rate (AFR) of more than 6%, much higher than the failure rates manufacturers predicted from extrapolating their short-term testing. The overall pattern was a few months of infant mortality, a two-year honeymoon of annual failure rates of a few percent, and then a jump up to the 6%–8% AFR range. Overall, hard disks in the Google study had less than a 75% chance of surviving a five-year tour of duty.

Interestingly, Google found no correlation between failure rate and two environmental factors that were formerly thought to be important: operating temperature and drive activity. The complete paper can be found at goo.gl/Y7Senk.

More recently, Backblaze, a cloud storage provider, has posted regular updates about its experience with various hard disk models at backblaze.com/blog. This data is 10 years more recent than the original Google study but suggests the same basic pattern: high infant mortality followed by a two- or three-year honeymoon and then a precipitous rise in annual failure rate. The absolute numbers are pretty close, too.
**Failure modes and metrics**

Hard disk failures typically stem from either platter surface defects (bad blocks) or mechanical failures. Drives attempt to transparently correct errors in the former category and remap the recovered data to a different portion of the disk. When block errors become visible at the operating system level (i.e., in the logs), that means data has already been lost. It’s a bad prognostic sign; pull the drive from service and replace it.

A disk’s firmware and hardware interface usually remain operable after a failure, and it can be entertaining to attempt to query the disk for details about what’s going on (see [this page](#)). However, disks are so cheap that it’s rarely worth your time to do this except perhaps as a learning exercise.

Drive reliability is often quoted by manufacturers in terms of mean time between failures (MTBF), denominated in hours. A typical value for an enterprise drive is around 1.2 million hours. However, MTBF is a statistical measure and should not be read to imply that an individual drive will run for 140 years before failing.

MTBF is defined as the inverse of AFR in the drive’s steady-state period—that is, after break-in but before wear-out. A manufacturer’s MTBF of 1.2 million hours corresponds to an AFR of 0.7% per year. This value is almost, but not quite, concordant with the AFR range observed by Google and Backblaze (1%–2%) during the honeymoon years of their sample drives’ lives.

Manufacturers’ MTBF values are probably accurate, but they are cherry-picked from the most reliable phase of each drive’s life. MTBF values should therefore be regarded as an upper bound on reliability; they do not predict your actual expected failure rate over the long term. (Our technical reviewer Jon Corbet refers to these as “reliability guaranteed not to exceed” values.) Based on the limited data quoted above, you might consider dividing manufacturers’ MTBFs by a factor of 7.5 or so to arrive at a more realistic estimate of five-year failure rates.

**Drive types**

Only two manufacturers of hard drives remain: Seagate and Western Digital. You may see a few other brands for sale, but they’re all ultimately made by these same two companies, both of which have been on decade-long acquisition binges.

Brands segment their hard disk offerings into a few general categories:

- **Value drives:** These products offer lots of storage at the lowest possible price point. Performance isn’t a priority, but it’s usually decent. Today’s low-end drives are often faster than the high-performance drives of five or ten years ago.

- **Mass-market performance drives:** These step-up products targeted at end users (often gamers) have higher spindle speeds and larger caches than those of their value equivalents. They perform notably better than value drives on most benchmarks. As
with value drives, firmware tuning emphasizes single-user access patterns such as large sequential reads and writes. The drives often run hot.

**NAS drives:** NAS stands for “network-attached storage,” but these drives are intended for use in all sorts of servers, RAID systems, and arrays—anywhere that multiple drives are housed and accessed together. They’re designed to be constantly on and working, and to balance performance, reliability, and low heat-emission.

Benchmarks that replicate stand-alone access patterns may not reveal much performance difference from value drives, but NAS drives typically handle multiple streams of independent operations more intelligently because of firmware tuning. NAS drives often have a longer warranty than value drives; their pricing is somewhere between that for value and performance drives.

**Enterprise drives:** “Enterprise” can mean a lot of things in the context of hard disks, but most commonly it means “expensive.” Here’s where you’ll find drives with non-SATA interfaces and uncommon features such as 10,000+ RPM spindle speeds. These are generally premium drives with long (often, five-year) warranties.

The differences among these drive categories are about half real and half marketing. All classes of drives work fine in all applications, but performance and reliability may vary. NAS drives are probably the best all-around choice for drives to keep on hand to fill a variety of potential needs.

Hard disks are commodity products, and one brand’s model of a given size, class, and spindle speed is much like another’s. These days, you need a dedicated qualification laboratory to make fine distinctions among competing drives, at least in terms of performance.

Reliability is another matter. The Google and Backblaze data demonstrate significant differences among models. The least reliable are an order of magnitude more likely to fail than the best. Unfortunately, there’s really no way to identify the turkeys until they’ve been sold for a year or two and have established a reputation in the real world.

That said, Hitachi (HGST, now part of Western Digital) deserves recognition as a particularly high-reliability brand. Over the last decade, its drives have consistently led the reliability charts. However, HGST-branded drives command a significant price premium over their competitors’ offerings.

No matter; even the best drives are relatively failure prone. There’s no escaping the need for backups and redundant storage when important data is at stake. Design your infrastructure with the assumption that drives will fail, then figure out how much an incrementally more reliable drive is worth within this context.

**Warranties and retirement**

Because hard drives are more likely to require warranty service than are other types of hardware, warranty length is an important purchasing consideration. The industry standard has shrunk to a
paltry two years, suspiciously close to the length of the average hard drive’s honeymoon period. The three-year warranty offered on many NAS drives is a significant advantage.

Hard disk exchanges under warranty are straightforward if you can demonstrate that drives fail a diagnostic test supplied by the manufacturer. Test programs typically run only under Windows and are intolerant of virtualization environments and of intervening connection hardware such as USB cradles. If your operations entail frequent drive exchanges, you may find it worthwhile to maintain a dedicated Windows machine as a drive testing station.

It usually pays to be aggressive in taking drives out of service, even if you can’t quite document that they are broken enough to be eligible for exchange under warranty. Even seemingly insignificant signs (e.g., funny noises or block errors within temporary files) are likely indications that a drive is nearing the end of its life.
**Solid state disks**

SSDs spread reads and writes across banks of flash memory cells, which are individually rather slow in comparison to modern hard disks. But because of parallelism, the SSD as a whole meets or exceeds the bandwidth of a traditional disk. The great strength of SSDs is that they continue to perform well when data is read or written at random, an access pattern that’s predominant in real-world use.

Storage device manufacturers like to quote sequential transfer rates for their products because the numbers are impressively high. But for traditional hard disks, these sequential numbers have almost no relationship to the throughput observed with random reads and writes. It pays to know your workloads. For access patterns that are in fact heavily sequential, hard disks can still be competitive with SSDs, especially when hardware costs are taken into consideration.

SSDs’ performance comes at a cost, however. Not only are they more expensive per gigabyte of storage than are hard disks, but they also introduce several new wrinkles and uncertainties into the storage equation. Anand Shimpi’s March 2009 article on SSD technology is a superb introduction to the promise and perils of the SSD. It can be found at [tinyurl.com/dexnbt](http://tinyurl.com/dexnbt).

**Rewritability limits**

Each page of flash memory in an SSD (typically 4KiB on current products) can be rewritten only a limited number of times (usually about 100,000, depending on the underlying technology). To limit the wear on any given page, the SSD firmware maintains a mapping table and distributes writes across all the drive’s pages. This remapping is invisible to the operating system, which sees the drive as a linear series of blocks. Think of it as virtual memory for storage.

The theoretical limits on the rewritability of flash memory are probably less an issue than they might initially seem. Just as a matter of arithmetic, you would have to stream 100 MB/s of data to a 500GB SSD for more than 15 continuous years to start running up against the rewrite limit. The more general question of long-term SSD reliability is as yet unanswered, however. We have a pretty good idea of how SSDs manufactured five years ago held up over time, but today’s products will no doubt behave differently.

**Flash memory and controller types**

SSDs are constructed from several types of flash memory. The main difference among the types has to do with how many bits of information are stored in each individual flash memory location. Single-level cells (SLC memories) store a single bit; they’re the fastest but most expensive option. Also common in the mix are multilevel cells (MLC) and triple-level cells (TLC).

SSD reviews lovingly describe these implementation details as a matter of course, but it’s not clear why buyers should care. Some SSDs are faster than others, but no particular hardware-related insight is needed to appreciate this fact. Standard benchmarks capture the performance differences quite well.
In theory, SLC flash memory has a reliability advantage over other types. In practice, reliability seems to have more to do with how well a drive’s firmware manages the memory and with how much memory the manufacturer has set aside for replacing cells that develop problems.

The controllers that coordinate SSD components are still evolving. Some are better than others, but these days all mainstream offerings tend to be respectable. If you want to invest time in scrutinizing SSD hardware, it’s usually more efficient to research the reputations of the flash memory controllers used to implement SSDs than to investigate the individual brands and models of SSD. SSD manufacturers are usually pretty open about the controllers they’re using. If they won’t tell you, reviewers certainly will.

Page clusters and pre-erasing

A further complication is that flash memory pages must be erased before they can be rewritten. SSDs handle this detail for you. However, erasing is a separate operation that is slower than writing. It’s also impossible to erase individual pages—clusters of adjacent pages (typically 128 pages or 512KiB) must be erased together. The write performance of an SSD can drop substantially when the pool of pre-erased pages is exhausted and the drive must recover pages on-the-fly to service ongoing writes.

Rebuilding a buffer of erased pages is harder than it might seem because filesystems designed for traditional hard disks do not actually erase data blocks they are no longer using. A storage device doesn’t know that the filesystem now considers a given block to be free; it knows only that long ago someone gave it data to store there. For an SSD to maintain its cache of pre-erased pages (and thus, its write performance), the filesystem must be capable of informing the SSD that certain pages are no longer needed. Support for this operation, known as TRIM, has finally become widespread among filesystems. On our example systems, the only filesystem that does not yet support TRIM is ZFS on Linux.

SSD reliability

A 2016 paper by Bianca Schroeder et al. (goo.gl/lzuX6c) summarized a vast set of SSD-related data from Google’s data centers. The main conclusions:

- Memory technology has no relationship to reliability. Reliability varies widely among models, but as with hard disks, it can be assessed only retrospectively.

- Most read errors occur at the bit level and are corrected through redundant storage coding. These “raw” (but correctable) read errors are common and expected. They occur on most SSD drives on most days of operation.

- The most common failure mode is to discover more bad bits in a block than can be fixed by the coding system. These errors are detectable but uncorrectable; they necessarily entail data loss.
- Even among the most reliable SSD models, 20% of drives experienced at least one uncorrectable read error. Among the least reliable models, 63%.

- Although both drive age and workload correlate with uncorrectable error rates, the correspondence is weak. In particular, the study found no evidence for the notion that older SSDs are ticking time bombs that asymptotically approach certain failure.

- Because uncorrectable errors are only marginally correlated to workload, the standard reliability figure quoted by manufacturers—the uncorrectable bit error rate, or UBER—is meaningless. Workload has little effect on the number of errors observed, so reliability should not be characterized as a rate.

The most notable of these findings is that unreadable blocks are common and that they typically occur in isolation. The usual scenario is for an SSD to report a block error but then continue to function normally.

See Chapter 28 for more information about setting up a comprehensive surveillance program.

Of course, unreliable storage devices are nothing new; backups and redundancy remain essential no matter what hardware you’re using. However, SSD failures are sneakier than those you might be accustomed to from dealing with hard disks. Unlike a hard disk, an SSD will rarely demand your attention by failing in some obvious and unambiguous way. SSDs need structured and systematic monitoring.

Errors develop over time regardless of a drive’s duty cycle, so SSDs are probably not a good choice for archival storage. And conversely, an isolated bad block is not an indication that an SSD has gone bad or is nearing the end of its useful life. In the absence of a larger pattern of failures, it’s fine to reformat such a drive and return it to service.
**Hybrid drives**

After spending many years in the vaporware category, SSHDs—hard disks with built-in flash memory caches—have become increasingly available. Current products are pitched at consumers.

The initialism SSHD stands for “solid state hybrid drive” and is something of a triumph of marketing, designed as it is to encourage confusion with SSDs. SSHDs are just traditional hard disks with some extras on the logic board; in reality, they’re about as “solid state” as the average dishwasher.

Benchmarks of current SSHD products have generally been unimpressive, even when the benchmarks attempt to emulate real-world access patterns. In large part, that’s because the current products often include only a token amount of flash memory cache.

Despite current SSHDs’ lackluster performance, the basic idea of multilevel caching is sound and has been well exploited in systems such as ZFS and Apple’s Fusion Drive. As the price of flash memory continues to fall, we anticipate that platter-based drives will continue to include more and more cache. Those products may or may not be sold explicitly as SSHDs.
Advanced Format and 4KiB blocks

For decades, the standard size of a disk block was fixed at 512 bytes. That’s too small to be practical from the perspective of most filesystems, so the filesystems themselves have long aggregated 512-byte sectors into page clusters of 1KiB to 8KiB that are read and written together.

Since no software that communicates with storage hardware actually has an interest in reading and writing data at 512-byte granularity, it’s inefficient and wasteful for the hardware to maintain such tiny sectors. Over the last decade, the storage industry has migrated to a new standard block size of 4KiB, known as Advanced Format. All modern storage devices use 4KiB sectors internally, although most of them continue to emulate 512-byte blocks from the perspective of clients.

There are currently three different “worlds” that a storage device can live in:

- **512n (or 512-native)** devices are the old ones that actually have 512-byte sectors. These devices are no longer manufactured, but of course there are still plenty of them out there in the real world. These drives know nothing about Advanced Format.

- **4Kn (or 4K-native)** devices are Advanced Format devices that have 4KiB sectors (or pages, in the case of SSDs) and that report their block size as 4KiB to the host computer. All interfacing hardware and all software that deals directly with the device must be aware of, and prepared to deal with, 4KiB blocks.

  4Kn is the wave of the future, but because it demands both hardware and software support, its adoption will be gradual. Enterprise drives with 4Kn interfaces started becoming available in 2014, but at this point you’re in no danger of encountering a 4Kn drive unless you explicitly order one.

- **512e (or 512-emulated)** devices use 4KiB blocks internally, but they report their sector size as 512 bytes to the host computer. Firmware in the device aggregates 512-byte block operations into operations on the actual 4KiB storage blocks.

The transition from 512n to 512e was completed in 2011. These two systems look essentially identical from the perspective of the host computer, so 512e devices work fine with old computers and old operating systems.

The one thing to know about 512e is that it’s sensitive to misalignment between filesystem page clusters and hardware disk blocks. Because the disk can only read or write 4KiB pages (despite its emulation of traditional 512-byte blocks), filesystem cluster boundaries and hard disk block boundaries should coincide. You wouldn’t want a 4KiB logical cluster to correspond to half of one 4KiB disk block and half of another—with that layout, the disk might have to read or write twice as many physical pages as it should to service a given number of logical clusters.
Since filesystems usually count off their clusters starting at the beginning of whatever storage is allocated to them, you can finesse the alignment issue by aligning disk partitions to a power-of-2 boundary that is large in comparison with the likely size of disk and filesystem pages (e.g., 64KiB). Partitioning tools on modern versions of Windows, Linux, and BSD automatically enforce such alignment. However, 512e disks that were mispartitioned on legacy systems can’t be transparently corrected; you’ll need to run an alignment utility to adjust the partition boundaries and physically move the data. Or, you can simply erase the device entirely and start over.
20.3 Storage hardware interfaces

These days, only a few interface standards are in common use. If a system supports several different interfaces, use the one that best meets your requirements for speed, redundancy, mobility, and price.
The SATA interface

Serial ATA, SATA, is the predominant hardware interface for storage. In addition to supporting high transfer rates (currently 6 Gb/s), SATA has native support for hot-swapping and (optional) command queueing, two features that finally make ATA a viable alternative to SAS in server environments.

SATA cables slide easily onto their mating connectors, but they can just as easily slide off. Cables with locking catches are available, but they’re a mixed blessing. On motherboards with six or eight SATA connectors packed together, it can be hard to disengage the locking connectors without a pair of needle-nosed pliers.

SATA also introduces an external cabling standard called eSATA. The cables are electrically identical to standard SATA, but the connectors are slightly different. You can add an eSATA port to a system that has only internal SATA connectors by installing an inexpensive converter bracket.

Be leery of external multidrive enclosures that have only a single eSATA port—some of these are smart (RAID) enclosures that require a proprietary driver, and the drivers rarely support UNIX or Linux. Others are dumb enclosures that have a SATA port multiplier built in. These are potentially usable on UNIX systems, but since not all SATA host adapters support port expanders, pay close attention to the compatibility information. Enclosures with multiple eSATA ports—one per drive bay—are always safe.
The PCI Express interface

The PCI Express (Peripheral Component Interconnect Express, abbreviated PCIe) backplane bus has been used on PC motherboards for more than a decade. It’s now the predominant standard for connecting all kinds of add-on circuit boards, even video cards.

As the SSD market developed, it became clear that even at 6 Gb/s, the speed of SATA interfaces would soon become inadequate to handle the fastest storage devices. Rather than assuming the traditional shape of a 2.5" laptop hard disk, high-end SSDs began to take the form of circuit boards that plugged directly into the system’s PCIe bus.

PCIe was attractive because of its flexible architecture and fast signaling rate. The version that is now mainstream, PCIe 3.0, has a signaling rate of 8 gigatransfers per second (GT/s). The actual throughput depends on how many signaling channels a device has; there can be as few as 1 or as many as 16. The widest devices can achieve more than 15 GB/s of throughput. (It’s not quite 16 GB/s because some of the bandwidth is consumed by signaling overhead. However, the amount of overhead is so small—about 1.5%—that it can safely be ignored.) The soon-to-debut PCIe 4.0 standard doubles the basic signaling rate to 16 GT/s.

When comparing PCIe to SATA, keep in mind that SATA’s speed of 6 Gb/s is quoted in gigabits per second. Full-width PCIe is actually more than 20 times faster than SATA.

The SATA standard is feeling the pressure. Unfortunately, the SATA ecosystem is constrained by past design choices and by the need to support existing cabling and connectors. It’s unlikely that the speed of SATA interfaces can be meaningfully improved over the next few years.

Instead, recent work has focused on attempting to unify SATA and PCIe at the level of interconnections. The M.2 standard for plug-in cards routes SATA, PCIe (with up to four data lanes), and USB 3.0 connectivity over a standard connector. One or two of these slots are now standard on laptop computers, and they can also be found on desktop systems.

M.2 cards are about an inch wide and can be up to about four inches long. They are thin, with only a few millimeters allowed on both sides for components.

U.2 is more recent tweak to the M.2 approach; it’s just starting to become available. Instead of USB, U.2 feature SAS connectivity in addition to SATA and PCIe.
The SAS interface

SAS stands for Serial Attached SCSI, the SCSI portion of which denotes the Small Computer System Interface, a generic data pipe that once connected many different types of peripherals. These days, USB has captured the market for peripheral connections and SCSI is found only in the form of SAS, an enterprise-level interface used to connect large numbers of storage devices.

Now that SAS and SCSI are largely synonymous, the vast history of different SCSI technologies dating back to 1986 serves mostly to create confusion. Operating systems further muddy the waters by filtering all disk access through a “SCSI subsystem” regardless of whether an actual SCSI device is involved or not. Our advice is to ignore all this history and consider SAS as its own system.

Like SATA, SAS is a point-to-point system: you plug a drive into a SAS port through a cable or direct-mount backplane. However, SAS allows “expanders” to connect multiple devices to a single host port. They’re analogous to SATA port multipliers, but whereas support for port multipliers is hit or miss, SAS expanders are always supported.

SAS currently operates at 12 Gb/s, twice the speed of SATA.

In past editions of this book, SCSI was the obvious interface choice for server applications. It offered the highest available bandwidth, out-of-order command execution (aka tagged command queueing), lower CPU utilization, easier handling of large numbers of storage devices, and access to the market’s most advanced hard drives.

The advent of SATA has removed or minimized most of these advantages, so SAS simply does not deliver the clear advantages that SCSI used to. SATA drives compete with (and in some cases, outperform) equivalent SAS disks in nearly every category. At the same time, both SATA devices and the interfaces and cabling used to connect them are cheaper and far more widely available.

SAS still holds a few trump cards:

- Manufacturers continue to use the SATA/SAS divide to stratify the storage market. To help justify premium pricing, the fastest and most reliable drives are still available only with SAS interfaces.

- SATA is limited to a queue depth of 32 pending operations. SAS can handle thousands.

- SAS can handle many storage devices (hundreds or thousands) on a single host interface. But keep in mind that all those devices share a single pipe to the host; you are still limited to 12 Gb/s of aggregate bandwidth.

The SAS vs. SATA debate may ultimately be moot because the SAS standard includes support for SATA drives. SAS and SATA connectors are similar enough that a single SAS backplane can
accommodate drives of either type. At the logical layer, SATA commands are simply tunneled over the SAS bus.

This convergence is an amazing technical feat, but the economic argument for it is less clear. The expense of a SAS installation is mostly in the host adapter, backplane, and infrastructure; the SAS drives themselves aren’t outrageously priced. Once you’ve invested in a SAS setup, you might as well stick with SAS from end to end. (On the other hand, perhaps the modest price premiums for SAS drives are a result of the fact that SATA drives can easily be substituted for them.)
The Universal Serial Bus (USB) is a popular option for connecting external hard disks. Current speeds are 4 Gb/s for USB 3.0 and up to 10 GB/s for USB 3.1. (The speed of USB 3.0 is often cited as 5 Gb/s, but because of mandatory encoding overhead, the actual transfer rate is more like 4 Gb/s.)

Both USB 3.0 and USB 3.1 are fast enough to accommodate all but the fastest SSDs streaming data at full speed. Watch out for USB 2.0, however; it tops out at 480 Mb/s, which is too slow to keep up with even a mechanical hard drive.

Storage devices themselves never come with native USB interfaces. External drives sold with these interfaces are invariably SATA drives with a protocol converter built into the enclosure. You can also buy these enclosures separately and install your choice of hard disks.

USB adapters are also available in the form of cradles and cable dongles. Cradles are particularly helpful when disks must be swapped out frequently: just yank out the old disk and pop in a new one.

USB thumb drives are perfectly legitimate storage devices. They present a block interface similar to that of any other disk, although throughput is typically mediocre. The underlying technology is similar to that of an SSD, but without some of the flourishes that give SSDs their superior speed and robustness.
20.4 Attachment and Low-Level Management of Drives

The way a disk is attached to the system depends on the interface. The rest is all mounting brackets and cabling. Fortunately, modern connection schemes are all pretty much idiot-proof.

See this page for more information about dynamic handling of devices.

SAS is a hot-pluggable interface, so it’s fine to plug in new drives without powering off the system or restarting it. The kernel should automatically recognize new devices and create device files for them. SATA interfaces can also theoretically support hot-plugging. However, the SATA specification does not require support for this feature, and most mass-market hardware does not implement it.

It’s fine to attempt hot-plugging a SATA drive to find out if hot-plugging works on a particular system. You won’t hurt anything. The worst that can happen is that the system ignores the drive.

Hot-plugging might seem like a neat trick that creates all sorts of options, such as the ability to swap out a bad drive with little or no software-side wrangling. However, it’s tricky to get the higher layers of the storage stack tuned to achieve these feats safely and reliably. We don’t describe the management of hot-plugging in this book.
Installation verification at the hardware level

After you install a new disk, check to make sure that the system acknowledges its existence at the lowest possible level. On a physical PC this is easy: the BIOS shows you a list of SATA and USB disks connected to the system. SAS disks may be included here as well if the motherboard supports them directly. If the system has a separate SAS interface card, you might need to invoke the BIOS setup for that card to see the disk inventory.

On cloud servers and systems that support hot-pluggable drives, you might have to do some sleuthing. Check the diagnostic output from the kernel as it probes for devices. For example, one of our test systems showed the following messages for an older SCSI disk attached to a BusLogic SCSI host adapter.

```
scsi0 : BusLogic BT-948
scsi : 1 host.
    Vendor: SEAGATE   Model: ST446452W   Rev: 0001
    Type: Direct-Access   ANSI SCSI revision: 02
Detected scsi disk sda at scsi0, channel 0, id 3, lun 0
scsi0: Target 3: Queue Depth 28, Asynchronous
SCSI device sda: hdwr sector=512 bytes. Sectors=91923356
    [44884 MB] [44.9 GB]
```

You may be able to review this information after the system has finished booting: look in your system log files. See the material starting on this page for more information about the handling of boot-time messages from the kernel.

Several commands can print out a list of the disks that the system is aware of. On Linux systems, the best option is usually `lsblk`, which is standard on all distributions. For more information, ask for model and serial numbers:

```
lsblk -o +MODEL,SERIAL
```

On FreeBSD, use `geom disk list`. 

Disk device files

A newly added disk is represented by device files in /dev. See this page for general information about device files.

All our example systems automatically create these files for you, but you still need to know where to look for the device files and how to identify the ones that correspond to your new device. Formatting the wrong device file is a rapid route to disaster.

Table 20.2 summarizes the device naming conventions for disks on our example systems. Instead of showing the abstract pattern according to which devices are named, Table 20.2 simply shows a typical example for the name of the system’s first disk.

<table>
<thead>
<tr>
<th>System</th>
<th>Whole disk</th>
<th>Partition</th>
</tr>
</thead>
<tbody>
<tr>
<td>Linux</td>
<td>/dev/sda</td>
<td>/dev/sda1</td>
</tr>
<tr>
<td>FreeBSD</td>
<td>/dev/ada0</td>
<td>/dev/ada0p1</td>
</tr>
</tbody>
</table>

Device names for whole disks comprise a basename that depends on the device driver and a sequence number or letter that differentiates disks from each other. For example, /dev/sda on Linux is the first drive managed by the sd driver. The next drive would be /dev/sdb, and so on. FreeBSD has different driver names and uses numbers instead of letters, but the pattern is the same.

Don’t ascribe too much significance to the driver names that show up in disk device files. Modern kernels funnel both SATA and SAS management through a generic SCSI layer, so don’t be surprised to see SATA disks masquerading as SCSI devices. Driver names also vary on cloud and virtualized systems; a virtual SATA disk may or may not have the same driver name as an actual SATA disk.

Device files for partitions add an additional decoration to the device file to indicate the partition number. Partition numbering normally starts at 1 rather than 0.
Ephemeral device names

Disk names are assigned in sequence as the kernel enumerates the various interfaces and devices on the system. Adding a disk can cause existing disks to change their names. In fact, even rebooting the system can sometimes cause name changes.

These facts suggest a couple of good rules for system administrators to follow:

- Never make changes to disks, partitions, or filesystems without verifying the identity of the disk you’re working on, even on a stable system.
- Never mention a disk device in any sort of configuration file, lest it change out from under you at some point in the future.

The latter issue is most notable when you are setting up the /etc/fstab file, which lists filesystems for the system to mount at boot time. It was once common to identify disk partitions by their device files in /etc/fstab, but this is no longer safe. See this page for some alternative approaches.

Linux has a couple of general ways around the “ephemeral names” issue. Subdirectories under /dev/disk list disks by various stable characteristics such as their manufacturer ID or connection information. These device names (which are really just links back to the top-level files in /dev) are stable, but they’re long and awkward.

At the level of filesystems and disk arrays, Linux uses both unique ID strings and text labels to persistently identify objects. In many cases, the existence of these long IDs is cleverly concealed so that you don’t have to deal with them directly.

parted -l lists the sizes, partition tables, model numbers, and manufacturers of every disk on the system.
Formatting and bad block management

People sometimes use the word “formatting” to mean “writing a partition table on a disk and setting up filesystems in the partitions.” But in this section, we use the word “formatting” to mean the more fundamental operation of setting up a disk’s media at the hardware level. We’d prefer to call the former operation “initializing,” but in the real world the terms are used more or less interchangeably, so you have to decode the meaning through context.

The formatting process writes address information and timing marks on the platters to delineate each sector. It also identifies bad blocks, imperfections in the media that result in areas that cannot be reliably read or written. All modern disks have bad block management built in, so neither you nor the driver need worry about managing defects. The drive firmware substitutes known-good blocks from an area of backup storage on the disk that is reserved for this purpose.

All hard disks come preformatted, and the factory formatting is at least as good as any formatting you can do in the field. It is best to avoid doing a low-level format if it’s not required. Don’t reformat new drives as a matter of course.

If you encounter read or write errors on a disk, first check for cabling, termination, and address problems, all of which can cause symptoms similar to those of a bad block. If after this procedure you are still convinced that the disk has defects, you might be better off replacing it with a new one rather than waiting long hours for a format to complete and hoping the problem doesn’t come back.

Bad blocks that manifest themselves after a disk has been formatted may or may not be automatically handled. If the drive is sure that the affected data can be reliably reconstructed, the newly discovered defect might be mapped out on the fly and the data rewritten to a new location. For more serious or less clearly recoverable errors, the drive aborts the read or write operation and reports the error back to the host operating system.

SATA disks are usually not designed to be formatted outside the factory. However, you might be able to obtain formatting software from the manufacturer, usually for Windows. Make sure the software matches the drive you plan to format and follow the manufacturer’s directions carefully. (On the other hand, at $100 for a 4TB drive, why bother?)

SAS disks format themselves in response to a standard command that you send from the host computer. The procedure for sending this command varies from system to system. On PCs, you can often send the command from the SAS controller’s BIOS. To issue the format command from within the operating system, use the `sg_format` command on Linux and the `camcontrol` command on FreeBSD.

Various utilities let you verify the integrity of a disk by writing random patterns to it and then reading them back. Thorough tests take a long time (hours) and unfortunately seem to be of little prognostic value. Unless you suspect that a disk is bad and are unable to simply replace it (or you bill by the hour), you can skip these tests. Barring that, let the tests run overnight. Don’t be
concerned about “wearing out” a disk with overuse or aggressive testing. Enterprise-class disks are designed for constant activity.
ATA secure erase

Since 2000, PATA and SATA disks have implemented a “secure erase” command that overwrites the data on the disk according to a method the manufacturer has determined to be secure against recovery efforts. Secure erase is NIST-certified for most needs. Under the U.S. Department of Defense categorization, it’s approved for use at security levels less than “secret.”

Why is this feature even needed? First, filesystems generally do no erasing of their own, so an `rm -rf *` of a disk’s data leaves everything intact and recoverable with software tools. It’s critically important to remember this fact when disposing of disks, whether their destination is eBay or the trash. (Now that most filesystems support the TRIM command to inform SSDs of blocks that are no longer needed by the system, this statement is not quite as true as it used to be. However, TRIM is advisory; an SSD is not required to erase anything in response.)

Second, even a manual rewrite of every sector on a traditional hard disk can leave magnetic traces that are recoverable by a determined attacker with access to a laboratory. Secure erase performs as many overwrites as are needed to eliminate these shadow signals. Magnetic remnants won’t be a serious concern for most sites, but it’s always nice to know that you’re not exporting your organization’s confidential data to the world at large. Some sites may have regulatory or business requirements that dictate how data is to be erased.

Finally, secure erase has the effect of resetting SSDs to their fully erased state. This reset can improve performance in cases in which the ATA TRIM command (the command to erase a block) cannot be issued, either because the filesystem used on the SSD does not know to issue it or because the SSD is connected through a host adapter or RAID interface that does not propagate TRIM.

The ATA secure erase command is password-protected at the drive level to reduce the risk of inadvertent activation. Therefore, you must set the password on a drive before invoking the command. Don’t bother to record the password, however; you can reset it at will. There is no danger of locking the drive.

Under Linux, you can use the `hdparm` command to activate secure erase:

```bash
$ sudo hdparm --user-master u --security-set-pass password /dev/disk
$ sudo hdparm --user-master u --security-erase password /dev/disk
```

The analogous FreeBSD command is `camcontrol`:

```bash
$ sudo camcontrol security disk -U user -s password -e password
```

The SAS world has no analog to ATA’s secure erase command, but the SCSI “format unit” command described under [Formatting and bad block management](#) is a reasonable alternative.
Many systems have a **shred** utility that attempts to securely erase the contents of individual files. Unfortunately, it relies on the assumption that a file’s blocks can be overwritten in place. This assumption is invalid in so many circumstances (any filesystem on any SSD, any logical volume that has snapshots, anything on ZFS or Btrfs) that **shred**'s general utility is questionable.

For sanitizing an entire PC system at once, another option is Darik’s Boot and Nuke (dban.org). This tool runs from its own boot disk, so it’s not a tool you’ll use every day. It is quite handy for decommissioning old hardware, however.
**hdparm and camcontrol: set disk and interface parameters**

The `hdparm` (Linux) and `camcontrol` (FreeBSD) commands can do more than just send secure erase commands. They give you a general way to interact with the firmware of SATA and SAS hard disks.

As tools that operate close to the hardware layer, these commands work properly only on nonvirtualized systems. On a traditional physical server, they are actually the best way to get information about the system’s disk devices (`hdparm -I` and `camcontrol devlist`); we don’t mention them elsewhere (e.g., in the “adding a disk” recipes at the start of this chapter) only because they don’t work on virtual systems.

`hdparm` comes from the prehistoric world of IDE and has gradually grown to include coverage of SATA and SCSI features. `camcontrol` started as a SCSI wrangling tool and has been extended to cover some SATA features. The syntaxes are different, but the tools cover approximately the same territory these days.

Among other things, these tools can set drive power options, enable or disable noise reduction options, set the read-only flag, and print detailed drive information.
Hard disk monitoring with SMART

Hard disks are fault-tolerant systems that use error-correction coding and intelligent firmware to hide their imperfections from the host operating system. In some cases, an uncorrectable error that the drive is forced to report to the OS is merely the latest event in a long crescendo of correctable but inauspicious problems. It would be nice to know about those omens before the crisis occurs.

SATA devices implement a detailed form of status reporting that is sometimes predictive of drive failures. This standard, called SMART, for “self-monitoring, analysis, and reporting technology,” exposes more than 50 operational parameters for investigation by the host computer.

The Google disk drive study mentioned on this page has been widely summarized in media reports as concluding that SMART data is not predictive of drive failure. That summary is not accurate. In fact, Google found that four SMART parameters were highly predictive of failure but that failure was not consistently preceded by changes in SMART values. Of failed drives in the study, 56% showed no change in the four most predictive parameters. On the other hand, predicting nearly half of failures sounds pretty good to us!

Those four sensitive SMART parameters are

- Scan error count
- Reallocation count
- Off-line reallocation count
- Number of sectors on probation

Those values should all be zero. According to the Google Labs study, a nonzero value in these fields raises the likelihood of failure within 60 days by a factor of 39, 14, 21, or 16, respectively.

To take advantage of SMART data, you need software that queries your drives to obtain it and then judges whether the current readings are sufficiently ominous to warrant administrator notification. Unfortunately, reporting standards vary by drive manufacturer, so decoding isn’t necessarily straightforward. Most SMART monitors collect baseline data and then look for sudden changes in the “bad” direction rather than interpreting absolute values. (According to the Google study, taking account of these “soft” SMART indicators in addition to the Big Four predicts 64% of all failures.)

The standard software for SMART wrangling is the smartmontools package from smartmontools.org. It’s installed by default on Red Hat, CentOS, and FreeBSD systems and is usually in the default package repository on other systems.

The smartmontools package consists of a smartd daemon that monitors drives continuously and a smartctl command you can use for interactive queries or for scripting. The daemon has a single configuration file, normally /etc.smartd.conf, which is extensively commented and includes plenty of examples.
SCSI has its own system for out-of-band status reporting, but unfortunately the standard is much less granular in this respect than is SMART. The smartmontools attempt to include SCSI devices in their schema, but the predictive value of the SCSI data is less clear.
20.5 The Software Side of Storage: Peeling the Onion

If you’re accustomed to plugging in a disk and having your Windows system ask if you want to format it, you may be a bit taken aback by the apparent complexity of storage management on UNIX and Linux systems. Why is it all so complicated?

To begin with, much of the complexity is optional. On UNIX and Linux systems with a window manager, you can log in to your system’s desktop, connect that same USB drive, and have much the same experience as on Windows. You’ll get a simple setup for personal data storage. If that’s all you need, you’re good to go.

As usual in this book, we’re primarily interested in enterprise-class storage systems: filesystems that are accessed by many users or processes (both local and remote) and that are reliable, high-performance, easy to back up, and easy to adapt to future needs. These systems require a bit more thought, and UNIX and Linux give you plenty to think about.
Elements of a storage system

*Exhibit A* shows a typical set of software components that can mediate between a raw storage device and its end users. The architecture shown in *Exhibit A* is for Linux, but other systems include similar features, although not necessarily in the same packages.

**Exhibit A: Storage management layers**

![Diagram of storage management layers](image)

The arrows in *Exhibit A* mean “can be built on.” For example, a Linux filesystem can be built on top of a partition, a RAID array, or a logical volume. It’s up to the administrator to construct a stack of modules that connect each storage device to its final application.

Sharp-eyed readers will note that the graph has a cycle, but real-world configurations should not loop. Linux allows RAID and logical volumes to be stacked in either order, but neither component should be used more than once (though it is technically possible to do this).

Here’s what the pieces in *Exhibit A* represent:

- **A storage device** is anything that looks like a disk. It can be a hard disk, a flash drive, an SSD, an external RAID array implemented in hardware, or even a network service that gives block-level access to a remote device. The exact hardware doesn’t matter, as long as the device allows random access, handles block I/O, and is represented by a device file.

- **A partition** is a fixed-size subsection of a storage device. Each partition has its own device file and acts much like an independent storage device. For efficiency, the same driver that handles the underlying device usually implements partitioning. Partitioning schemes consume a few blocks at the start of the device to record the ranges of blocks in each partition.

- **Volume groups** and **logical volumes** are associated with logical volume managers (LVMs). These systems aggregate physical devices to form pools of storage called volume groups. An administrator can then subdivide this pool into logical volumes in

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Volume groups and logical volumes are associated with logical volume managers (LVMs). These systems aggregate physical devices to form pools of storage called volume groups. An administrator can then subdivide this pool into logical volumes in
much the same way that disks can be divided into partitions. For example, a 6TB disk and a 2TB disk could be aggregated into an 8TB volume group and then split into two 4TB logical volumes. At least one volume would include data blocks from both hard disks.

Since the LVM adds a layer of indirection between logical and physical blocks, it can freeze the logical state of a volume simply by making a copy of the mapping table. Therefore, logical volume managers often have some kind of a “snapshot” feature. Writes to the volume are then directed to new blocks, and the LVM keeps both the old and new mapping tables. Of course, the LVM has to store both the original image and all modified blocks, so it can eventually run out of space if a snapshot is never deleted.

- A RAID array (a redundant array of inexpensive/independent disks) combines multiple storage devices into one virtualized device. Depending on how you set up the array, this configuration can increase performance (by reading or writing disks in parallel), increase reliability (by duplicating or parity-checking data across multiple disks), or both. RAID can be implemented by the operating system or by various types of hardware.

As the name suggests, RAID is typically conceived of as an aggregation of bare drives, but modern implementations let you use as a component of a RAID array anything that acts like a disk.

- A filesystem mediates between the raw bag of blocks presented by a partition, RAID array, or logical volume and the standard filesystem interface expected by programs: paths such as /var/spool/mail, UNIX file types, UNIX permissions, etc. The filesystem determines where and how the contents of files are stored, how the filesystem namespace is represented and searched on disk, and how the system is made resistant to (or recoverable from) corruption.

Most storage space ends up as part of a filesystem, but on some systems (not current versions of Linux), swap space and database storage can potentially be slightly more efficient without “help” from a filesystem. The kernel or database imposes its own structure on the storage, rendering the filesystem unnecessary.

If it seems to you that this taxonomy has a few too many little components that simply implement one block storage device in terms of another, you’re in good company. The trend over the last few years has been toward consolidating these components to increase efficiency and remove duplication. Although logical volume managers did not originally function as RAID controllers, most have absorbed some RAID-like features (notably, striping and mirroring).

On the cutting edge today are systems that combine a filesystem, a RAID controller, and an LVM system all in one tightly integrated package. ZFS was the earliest example, but the Btrfs
filesystem for Linux has similar design goals. We have lots more to say about ZFS and Btrfs starting on this page. (Spoiler alert: if you can use one of these systems, you probably should.)
The Linux device mapper

For simplicity, we omitted a central component of the Linux storage stack from Exhibit A: the device mapper. This is a protean little beastie that has fingers inserted in multiple contexts, prime examples being the implementation of LVM2, the implementation of filesystem layers for containerization (see Chapter 25), and the implementation of whole-disk encryption (search the web for LUKS).

The device mapper abstracts the idea of one block device being built on a collection of other block devices. Given a mapping table of devices, it implements the ongoing translation among them and routes each block to its appropriate home.

For the most part, the device mapper is part of the implementation of Linux storage and not something you’ll deal with directly. However, you’ll see its traces whenever you access devices under /dev/mapper. You can also set up your own mapping tables with the dmsetup command, although cases in which you might need to do that are relatively rare.

In the next sections, we look in more detail at the layers involved in storage configuration: partitioning, RAID, logical volume management, and filesystems.
20.6 Disk partitioning

Partitioning and logical volume management are both ways of dividing up a disk (or pool of
disks, in the case of LVM) into separate chunks of known size. Linux and FreeBSD support both
of these methods.

Traditionally, partitioning was the lowest possible level of disk management, and only disks
could be partitioned. You could put individual disk partitions under the control of a RAID
controller or logical volume manager, for example, but you couldn’t then partition the resulting
logical volumes or RAID volumes.

The rule that only disks can be partitioned is increasingly being waived in favor of a more
general model in which disks, partitions, LVM pools, and RAID arrays can be derived from one
another in any order or combination. From the standpoint of software architecture, this is
beautiful and elegant. But from the standpoint of practicality, it has the unfortunate side effect of
implying that there’s some valid reason to partition entities other than disks.

In fact, partitioning is less desirable than logical volume management in most respects. It’s
coarse and brittle and lacks features such as snapshot management. Partitioning decisions are
difficult to revise later. The only notable advantages of partitioning over logical volume
management are its simplicity and the fact that Windows and PC BIOSs understand and expect
it. A few versions of UNIX that run on proprietary hardware have done away with partitioning
altogether, and nobody on those systems seems to miss it.

Both partitions and logical volumes make backups easier, prevent users from poaching each
other’s disk space, and confine potential damage from runaway programs. All systems have a
root “partition” that includes / and most of the local host’s configuration data. In theory,
everything needed to bring the system up to single-user mode is part of the root partition. Various
subdirectories (most commonly /var, /usr, /tmp, /share, and /home) can be broken out into their
own partitions or volumes. Most systems also have at least one swap area.

Opinions differ on the best way to divide up disks, as do the defaults used by various systems.
Most setups are relatively simple. Exhibit B illustrates a traditional partitions-and-filesystems
schema as it might be found on a couple of data disks on a Linux system. (The boot disk is not
shown.)

Exhibit B: Traditional data disk partitioning scheme (Linux device names)
Here are some general points to guide you:

- In the distant past, it was sometimes useful to have a backup root device that you could boot to if something went wrong with the normal root partition. These days, a bootable USB thumb drive or an OS installation DVD is a better recovery option for most systems. Backup root partitions are more trouble than they’re worth.

- Putting `/tmp` on a separate filesystem limits temporary files to a finite size and saves you from having to back them up. Some systems use a memory-based filesystem to hold `/tmp` for performance reasons. The memory-based filesystems are still backed by swap space, so they work well in a broad range of situations.

- Since log files are kept in `/var/log`, it’s a good idea for either `/var` or `/var/log` to be a separate disk partition. Leaving `/var` as part of a small root partition makes it easy to fill the root and bring the machine to a halt.

- It’s useful to put users’ home directories on a separate partition or volume. Even if the root partition is corrupted or destroyed, user data has a good chance of remaining intact. Conversely, the system can continue to operate even after a user’s misguided shell script fills up `/home`.

- Splitting swap space among several physical disks can potentially increase performance, although with today’s cheap RAM it’s usually better not to swap at all. This technique works for filesystems, too; put the busy ones on different disks. See this page for notes on this subject.

- As you add memory to your machine, also add swap space. See this page for more information about virtual memory.

- Try to cluster quickly changing information on a few partitions that are backed up frequently.

- The Center for Internet Security publishes configuration guidelines for a variety of operating systems at www.cisecurity.org/cis-benchmarks. They are “benchmarks” in the sense of being best practices. The documents include helpful recommendations for partitioning and filesystem layout.
**Traditional partitioning**

Systems that support partitions implement them by writing a “label” at the beginning of the disk to define the range of blocks included in each partition. The exact details vary; the label must often coexist with other startup information (such as a boot block), and it often contains extra information such as a name or unique ID that identifies the disk as a whole.

The device driver responsible for representing the disk reads the label and uses the partition table to calculate the physical location of each partition. Typically, one device file represents each partition and an additional device file represents the disk as a whole.

Despite the universal availability of logical volume managers, some situations still require or benefit from traditional partitioning.

- Only two partitioning schemes are used these days: MBR and GPT. We discuss the details of both schemes in the next sections.

- On PC hardware, the boot disk must have a partition table. Systems manufactured before 2012 usually require MBR, and some new systems require GPT. Most new systems support both.

- Installing an MBR or GPT partition table makes a disk comprehensible to Windows, even if the contents of the individual partitions are not. Though you may have no particular plans to interoperate with Windows, consider the ubiquity of Windows, the prevalence of virtual machines, and the portability of hard disks.

- Partitions have a defined location on the disk, so they guarantee locality of reference. Logical volumes do not (at least, not by default). In most cases, this fact isn’t terribly important. However, short seeks are faster than long seeks on mechanical hard disks, and the throughput of a disk’s outer cylinders (those containing the lowest-numbered blocks) can exceed the throughput of its inner cylinders by 30% or more.

- RAID systems (see this page) use disks or partitions of matched size. A given RAID implementation might accept entities of different sizes, but it will probably use only the block ranges that all devices have in common. Rather than letting extra space go to waste, you can isolate it in a separate partition. If you do this, however, use the spare partition for data that is infrequently accessed; otherwise, traffic on the partition will degrade the performance of the RAID array.
**MBR partitioning**

MBR (Master Boot Record) partitioning is an old Microsoft standard that dates back to the 1980s. It’s a cramped and ill-conceived format that can’t support disks larger than 2TB. Who knew disks could ever get that big?

MBR offers no advantages over GPT except that it’s the only format from which old PC hardware can boot Windows. Unless you’re forced by circumstances to use MBR partitions, you typically don’t want them. Unfortunately, MBR is still a common default setup for many distributions’ installers.

The MBR label occupies a single 512-byte disk block, most of which is consumed by boot code. Only enough space remains to define four partitions. These are termed “primary” partitions because they are defined directly in the MBR.

You can, in theory, define one of the primary partitions to be an “extended” partition, which means that it contains its own subsidiary partition table. Unfortunately, extended partitions have been known to cause a variety of subtle problems. It’s best to avoid them in these twilight years of MBR.

The Windows partitioning system lets one partition be marked “active.” Boot loaders look for the active partition and try to load the operating system from it.

Each partition also has a one-byte type attribute that is supposed to signal the partition’s contents. Generally, the codes represent either filesystem types or operating systems. These codes are not centrally assigned, but some common conventions have evolved. They are summarized by Andries E. Brouwer at [goo.gl/ATi3](http://goo.gl/ATi3).

The MS-DOS command that partitioned hard disks was called `fdisk`. Most operating systems that support MBR-style partitions have adopted this name for their own partitioning commands, but there are many variations among `fdisks`. Windows itself has moved on: the command-line tool in recent versions is called `diskpart`. Windows also has a partitioning GUI that’s available through the Disk Management plug-in of `mmc`.

It does not matter whether you partition a disk with Windows or some other operating system. The end result is the same.
GPT: GUID partition tables

Intel’s extensible firmware interface (EFI) project replaced the rickety conventions of PC BIOSs with a more modern and functional architecture. EFI firmware is now standard for new PC hardware, and EFI’s partitioning scheme has gained universal support among operating systems.

EFI has more recently become UEFI, a “unified” EFI effort supported by multiple vendors. However, EFI remains the more common term in general use. UEFI and EFI are essentially interchangeable.

The EFI partitioning scheme, known as a “GUID partition table” or GPT, removes the obvious weaknesses of MBR. It defines only one kind of partition (no more “logical partitions in the extended partition”), and you can create arbitrarily many of them. Each partition has a type specified by a 16-byte ID code (a globally unique ID, or GUID) that requires no central arbitration.

Significantly, GPT retains primitive compatibility with MBR-based systems by dragging along an MBR as the first block of the partition table. This “fakie” MBR makes the disk look like it’s occupied by one large MBR partition (at least, up to the 2TB limit of MBR). It isn’t useful per se, but the hope is that the decoy MBR will at least prevent naïve systems from attempting to reformat the disk.

Versions of Windows from the Vista era forward support GPT disks for data, but only systems with EFI firmware can boot Windows from them. Linux and its GRUB boot loader have fared better: GPT disks are supported by the OS and bootable on any system. Intel-based macOS systems use both EFI and GPT partitioning.

Although GPT has already been well accepted by operating system kernels, many disk management utilities are unmaintained and lack support for it. Make sure that any utility you run on a GPT disk actually supports GPT.
Linux partitioning

Linux systems give you several options for partitioning, which makes for treacherous terrain, given that some of the offerings are not GPT-aware. Default to parted, a command-line tool that understands several label formats (including Solaris’s native one) and can move and resize partitions in addition to simply creating and deleting them. A GUI version, gparted, runs under GNOME.

In general, we recommend gparted over parted. Both are simple, but with gparted you can specify the size of the partitions you want instead of specifying the starting and ending block ranges. For partitioning the boot disk, most distributions’ graphical installers are the best option since they typically suggest a partitioning plan that works well with that particular distribution’s layout.
FreeBSD partitioning

Like Linux, FreeBSD has several partitioning tools. Ignore all except `gpart`. The others exist only to lure you into making some kind of terrible mistake.

The mysterious “geoms” you’ll see referred to in the `gpart` man page (and in other storage-related contexts on FreeBSD) are FreeBSD’s abstraction of storage devices. Not all geoms are disk drives, but all disk drives are geoms, so you can use a generic disk name such as `ada0` wherever a geom is called for.

The FreeBSD “adding a disk” recipe on this page uses `gpart` to configure the partition table on a new disk.
20.7 Logical Volume Management

Imagine a world in which you don’t know exactly how large a partition needs to be. Six months after creating the partition, you discover that it is much too large, but that a neighboring partition doesn’t have enough space. Sound familiar? A logical volume manager lets you reallocate space dynamically from the greedy partition to the needy partition.

Logical volume management is essentially a supercharged and abstracted version of disk partitioning. It groups individual storage devices into “volume groups.” The blocks in a volume group can then be allocated to “logical volumes,” which are represented by block device files and act like disk partitions.

However, logical volumes are more flexible and powerful than disk partitions. Here are some of the magical operations a volume manager lets you carry out:

- Move logical volumes among different physical devices
- Grow and shrink logical volumes on the fly
- Take copy-on-write “snapshots” of logical volumes
- Replace on-line drives without interrupting service
- Incorporate mirroring or striping in your logical volumes

The components of a logical volume can be put together in various ways. Concatenation keeps each device’s physical blocks together and lines the devices up one after another. Striping interleaves the components so that adjacent virtual blocks are actually spread over multiple physical disks. By reducing single-disk bottlenecks, striping can often result in higher bandwidth and lower latency.

If you’ve had some prior exposure to RAID (see the section starting on this page), you might find striping reminiscent of RAID 0. LVM implementations of striping tend to be more flexible than RAID, though. For example, they may automatically optimize striping or allow devices of different sizes to be striped, even if striping won’t actually happen 100% of the time. The line between LVM and RAID has become blurry indeed, and even parity schemes like RAID 5 and RAID 6 are making regular appearances in volume managers.
Linux logical volume management

Linux’s volume manager, called LVM2, is essentially a clone of HP-UX’s volume manager, which is itself based on software by Veritas. The commands for the two systems are essentially identical. Table 20.3 summarizes the LVM command set.

Table 20.3: LVM commands in Linux

<table>
<thead>
<tr>
<th>Entity</th>
<th>Operation</th>
<th>Command</th>
</tr>
</thead>
<tbody>
<tr>
<td>Physical volume</td>
<td>Create</td>
<td>pvcreate</td>
</tr>
<tr>
<td></td>
<td>Inspect</td>
<td>pvdisplay</td>
</tr>
<tr>
<td></td>
<td>Modify</td>
<td>pvchange</td>
</tr>
<tr>
<td></td>
<td>Check</td>
<td>pvck</td>
</tr>
<tr>
<td>Volume group</td>
<td>Create</td>
<td>vgcreate</td>
</tr>
<tr>
<td></td>
<td>Modify</td>
<td>vgchange</td>
</tr>
<tr>
<td></td>
<td>Extend</td>
<td>vgeextend</td>
</tr>
<tr>
<td></td>
<td>Inspect</td>
<td>vgdisplay</td>
</tr>
<tr>
<td></td>
<td>Check</td>
<td>vgck</td>
</tr>
<tr>
<td></td>
<td>Enable</td>
<td>vgscan</td>
</tr>
<tr>
<td>Logical volume</td>
<td>Create</td>
<td>lvcreate</td>
</tr>
<tr>
<td></td>
<td>Modify</td>
<td>lvchange</td>
</tr>
<tr>
<td></td>
<td>Resize</td>
<td>lvresize</td>
</tr>
<tr>
<td></td>
<td>Inspect</td>
<td>lvdisplay</td>
</tr>
</tbody>
</table>

The top-level architecture of LVM is that individual disks and partitions (physical volumes) are gathered into storage pools called volume groups. Volume groups are then subdivided into logical volumes, which are the block devices that hold filesystems.

A physical volume needs to have an LVM label applied with pvcreate. Applying such a label is the first step to accessing the device through the LVM. In addition to bookkeeping information, the label includes a unique ID to identify the device.

“Physical volume” is a somewhat misleading term because physical volumes need not have a direct correspondence to physical devices. They can be disks, but they can also be disk partitions or RAID arrays. LVM doesn’t care.

You can control LVM with either a large group of simple commands (the ones listed in Table 20.3) or with the single lvm command and its various subcommands. These options are essentially identical; in fact, the individual commands are just links to lvm, which looks to see how it’s been called to know how to behave. man lvm is a good introduction to the system and its tools.
A Linux LVM configuration proceeds in a few distinct phases:

- Creating (defining, really) and initializing physical volumes
- Adding the physical volumes to a volume group
- Creating logical volumes on the volume group

LVM commands start with letters that make it clear at which level of abstraction they operate: `pv` commands manipulate physical volumes, `vg` commands manipulate volume groups, and `lv` commands manipulate logical volumes. A few commands with the prefix `lvm` (e.g., `lvmchange`) operate on the system as a whole.

In the following example, we set up a 1TB hard disk (/dev/sdb) for use with LVM and create a logical volume. We assume that the disk has been partitioned as described on this page, with all space being assigned to a single partition, /dev/sdb1. We could omit the partitioning step entirely and just use the raw disk as our physical device, but there is no performance benefit to doing so. Partitioning makes the disk comprehensible to the broadest variety of software and operating systems.

The first step is to label the `sdb1` partition as an LVM physical volume:

```
$ sudo pvcreate /dev/sdb1
Physical volume "/dev/sdb1" successfully created
```

Our physical device is now ready to be added to a volume group:

```
$ sudo vgcreate DEMO /dev/sdb1
Volume group "DEMO" successfully created
```

Although we’re using only a single physical device in this example, we could of course add additional devices. To step back and examine our handiwork, we use the `vgdisplay` command:
A PE is a physical extent, the allocation unit according to which the volume group is subdivided.

The final steps are to create the logical volume within DEMO and then to create a filesystem within that volume. We make the logical volume 100GB in size:

$ sudo lvcreate -L 100G -n web1 DEMO
Logical volume "web1" created

Most of LVM’s interesting options live at the logical volume level. That’s where striping, mirroring, and contiguous allocation would be requested if we were using those features.

We can now access the volume through the device /dev/DEMO/web1. We discuss filesystems in general starting on this page, but here is a quick overview of creating an ext4 filesystem so that we can demonstrate a few additional LVM tricks.

$ sudo mkfs /dev/DEMO/web1
...
$ sudo mkdir /mnt/web1
$ sudo mount /dev/DEMO/web1 /mnt/web1

Volume snapshots
You can create copy-on-write duplicates of any LVM logical volume, whether or not it contains a filesystem. This feature is handy for creating a quiescent image of a filesystem to be backed up elsewhere, but unlike ZFS and Btrfs snapshots, LVM2 snapshots are unfortunately not very useful as a general method of version control.
The problem is that logical volumes are of fixed size. When you create one, storage space is allocated for it up front from the volume group. A copy-on-write duplicate initially consumes no space, but as blocks are modified, the volume manager must find space in which to store both the old and new versions. This space for modified blocks must be set aside when you create the snapshot, and like any LVM volume, the allocated storage is of fixed size.

Note that it does not matter whether you modify the original volume or the snapshot (which by default is writable). Either way, the cost of duplicating the blocks is charged to the snapshot. Snapshots’ allocations can be pared away by activity on the source volume even when the snapshots themselves are idle.

If you do not allocate as much space for a snapshot as is consumed by the volume of which it is an image, you can potentially run out of space in the snapshot. That’s more catastrophic than it sounds because the volume manager then has no way to maintain a coherent image of the snapshot; additional storage space is required just to keep the snapshot the same. The result of running out of space is that LVM stops maintaining the snapshot, and the snapshot becomes corrupted.

So, as a matter of practice, LVM snapshots should be either short-lived or as large as their source volumes. So much for “lots of cheap virtual copies.”

To create /dev/DEMO/web1-snap as a snapshot of /dev/DEMO/web1, we would use the following command:

```
$ sudo lvcreate -L 100G -s -n web1-snap DEMO/web1
Logical volume "web1-snap" created.
```

Note that the snapshot has its own name and that the source of the snapshot must be specified as volume_group/volume.

In theory, /mnt/web1 should really be unmounted first to ensure the consistency of the filesystem. In practice, ext4 protects us against filesystem corruption, although we might lose a few of the most recent data block updates. This is a perfectly reasonable compromise for a snapshot used as a backup source.

To check on the status of your snapshots, run lvdisplay. If lvdisplay tells you that a snapshot is “inactive,” that means it has run out of space and should be deleted. There’s little you can do with a snapshot once it reaches this point.

**Filesystem resizing**

Filesystem overflows are more common than disk crashes, and one advantage of logical volumes is that they’re much easier to juggle and resize than are hard partitions. We have experienced everything from servers used for personal video storage to departments full of email pack rats.

The logical volume manager doesn’t know anything about the contents of its volumes, so you must do your resizing at both the volume and filesystem levels. The order depends on the specific
operation. Reductions must be filesystem-first, and enlargements must be volume-first. Don’t memorize these rules: just think about what’s actually happening and use common sense.

Suppose that in our example, /mnt/web1 has grown more than we predicted and needs another 100GB of space. We first check the volume group to be sure additional space is available.

```
$ sudo vgdisplay DEMO
--- Volume group ---
 VG Name       DEMO
 System ID
 Format   lvm2
 Metadata Areas  1
 Metadata Sequence No  4
 VG Access    read/write
 VG Status    resizable
 Open LV      1
 Max PV       0
 Cur PV       1
 Act PV       1
 VG Size     1000.00 GiB
 PE Size     4.00 MiB
 Total PE    255999
 Alloc PE / Size      51200 / 200.00 GiB
 Free PE / Size       204799 / 800.00 GiB
 VG UUID      n26rxj-X5HN-x4nv-rdnM-7AWe-QQ21-EdDwEO
```

Note that 200GB of space has been consumed, 100GB for the original filesystem and 100GB for the snapshot. However, plenty of space is still available. We unmount the filesystem and use `lvresize` to add space to the logical volume.

```
$ sudo umount /mnt/web1
$ sudo lvchange -an DEMO/web1
$ sudo lvresize -L +100G DEMO/web1
Size of logical volume DEMO/web1 changed from 100.00 GiB (25600 extents) to 200.00 GiB (51200 extents).
Logical volume DEMO/web1 successfully resized.
$ sudo lvchange -ay DEMO/web1
```

The `lvchange` commands are needed to deactivate the volume for resizing and to reactivate it afterwards. This part is needed only because an existing snapshot of web1 remains from our previous example. After the resize operation, the snapshot will “see” the additional 100GB of allocated space, but since the filesystem it contains is only 100GB in size, the snapshot will still be usable.

We can now resize the filesystem with `resize2fs`. (The 2 comes from the original ext2 filesystem, but the command supports all versions of ext.) Since `resize2fs` can determine the size of the new filesystem from the volume, we don’t need to specify the new size explicitly. We would have to do so when shrinking the filesystem:
That’s it! Examining the output of `df` again shows the changes:

```
$ sudo mount /dev/DEMO/web1 /mnt/web1
$ df -h /mnt/web1
Filesystem Size Used Avail Use% Mounted on
/devmapper/DEMO-web1  197G  60M  187G  1%  /mnt/web1
```

Commands for resizing other filesystems work similarly. For XFS filesystems (the default on Red Hat and CentOS systems), use `xfs_growfs`; for UFS filesystems (the default on FreeBSD), use `growfs`. XFS filesystems must be mounted to be expanded. As the names of these commands suggest, XFS and UFS filesystems can be expanded but not made smaller. If you need to remove space, you’ll need to copy the filesystem’s contents to a new, smaller filesystem.

It’s worth noting that “disks” you allocate and attach to virtual machines in the cloud are essentially logical volumes, although the volume manager itself lives elsewhere in the cloud. These volumes are usually resizable through the cloud provider’s management console or command-line utility.

The procedure for resizing cloud filesystems is much the same as the one outlined above, but keep in mind that because these virtual devices impersonate disk drives, they probably have partition tables. You’ll need to resize on three separate layers: at the cloud provider level, at the partition level, and at the filesystem level.
FreeBSD logical volume management

FreeBSD has a full-fledged logical volume manager of its own. Previous versions were known by the name Vinum, but now that the system has been rewritten to conform to FreeBSD’s generalized geom architecture for storage devices, the name has been changed to GVinum. Like LVM2, GVinum implements a variety of RAID types.

FreeBSD has recently put a lot of effort into ZFS support, and although GVinum has not been officially deprecated, developers’ public comments suggest that ZFS is the recommended approach for logical volume management and RAID going forward. Accordingly, we do not discuss GVinum here. ZFS is covered starting on this page.
20.8 RAID: **REDUNDANT ARRAYS OF INEXPENSIVE DISKS**

Even with backups, the consequences of a disk failure on a server can be disastrous. RAID, “redundant arrays of inexpensive disks,” is a system that distributes or replicates data across multiple disks. RAID is sometimes glossed as “redundant arrays of independent disks,” too. Both versions are historically accurate.

RAID not only helps avoid data loss but also minimizes the downtime associated with hardware failures (often to zero) and potentially increases performance.

RAID can be implemented by dedicated hardware that presents a group of hard disks to the operating system as a single composite drive. It can also be implemented simply by the operating system’s reading or writing multiple disks according to the rules of RAID.
Software vs. hardware RAID

Because the disks themselves are always the most significant bottleneck in a RAID implementation, there is no reason to assume that a hardware-based implementation of RAID will necessarily be faster than a software- or OS-based implementation. Hardware RAID has been predominant in the past for two main reasons: lack of software alternatives (no direct OS support for RAID) and hardware’s ability to buffer writes in some form of nonvolatile memory.

The latter feature does improve performance because it makes writes appear to complete instantaneously. It also protects against a potential corruption issue called the “RAID 5 write hole,” which we describe in more detail starting on this page. But beware: many of the common “RAID cards” sold for PCs have no nonvolatile memory at all; they are just glorified SATA interfaces with some RAID software on-board. RAID implementations on PC motherboards fall into this category as well. You’re better off using the RAID features in Linux or FreeBSD on these systems. (Or better yet, use ZFS or Btrfs.)

We have experienced a disk controller failure on an important production server. Although the data was replicated across several physical drives, a faulty hardware RAID controller destroyed the data on all disks. A lengthy and ugly restore process ensued. The rebuilt server now relies on the kernel’s software to manage its RAID environment, removing the possibility of another RAID controller failure.
RAID levels

RAID can do two basic things. First, it can improve performance by “striping” data across multiple drives, thus allowing several drives to work simultaneously to supply or absorb a single data stream. Second, it can replicate data across multiple drives, decreasing the risk associated with a single failed disk.

Replication assumes two basic forms: mirroring, in which data blocks are reproduced bit-for-bit on several different drives, and parity schemes, in which one or more drives contain an error-correcting checksum of the blocks on the remaining data drives. Mirroring is faster but consumes more disk space. Parity schemes are more disk-space-efficient but have lower performance.

RAID is traditionally described in terms of “levels” that specify the exact details of the parallelism and redundancy implemented by an array. The term is perhaps misleading because “higher” levels are not necessarily “better.” The levels are simply different configurations; use whichever versions suit your needs.

In the following illustrations, numbers identify stripes and the letters a, b, and c identify data blocks within a stripe. Blocks marked p and q are parity blocks.

“Linear mode,” also known as JBOD (for “just a bunch of disks”) is not even a real RAID level. And yet, every RAID controller seems to implement it. JBOD concatenates the block addresses of multiple drives to create a single, larger virtual drive. It has no data redundancy or performance benefit. These days, JBOD functionality is best achieved through a logical volume manager rather than a RAID system.

RAID level 0 increases performance. It combines two or more drives of equal size, but instead of stacking them end-to-end, it stripes data alternately among the disks in the pool. Sequential reads and writes are therefore spread among several disks, decreasing write and access times.

Note that RAID 0 has reliability characteristics that are significantly inferior to separate disks. A two-drive array has roughly double the annual failure rate of a single drive, and so on.

RAID level 1 is colloquially known as mirroring. Writes are duplicated to two or more drives simultaneously. This arrangement makes writes slightly slower than they would be on a single drive. However, it offers read speeds comparable to RAID 0 because reads can be farmed out among the several duplicate disk drives.
RAID levels 1+0 and 0+1 are stripes of mirrors or mirrors of stripes. Logically, they are concatenations of RAID 0 and RAID 1, but many controllers and software implementations support them directly. The goal of both modes is to simultaneously obtain the performance of RAID 0 and the redundancy of RAID 1. These configurations need at least four devices.

RAID level 5 stripes both data and parity information, adding redundancy while simultaneously improving read performance. In addition, RAID 5 is more efficient in its use of disk space than is RAID 1. If an array has N drives (at least three are required), N–1 of them can store data. The space-efficiency of RAID 5 is therefore at least 67%, whereas that of mirroring cannot be higher than 50%.

RAID level 6 is similar to RAID 5 with two parity disks. A RAID 6 array can withstand the complete failure of two drives without losing data. RAID 6 requires at least four devices.
RAID levels 2, 3, and 4 are defined but rarely deployed. Logical volume managers usually include both striping (RAID 0) and mirroring (RAID 1) features.

As RAID systems, logical volume managers, and filesystem all rolled into one, ZFS and Btrfs support striping, mirroring, and configurations similar to RAID 5 and RAID 6. See this page for more details on these options.

Linux supports both ZFS and Btrfs, though you might have to install ZFS separately. Btrfs’s RAID 5 and RAID 6 support is not officially ready for production use.

For simple striped and mirrored configurations outside the context of one of these filesystems, Linux gives you a choice between a dedicated RAID system (md; see this page) and the logical volume manager, LVM. The LVM approach is perhaps more flexible, but the md approach may be a bit more rigorously predictable. If you opt for md, you can still use LVM to manage the space on the RAID volume. For RAID 5 and RAID 6, you must use md to implement software RAID.

ZFS is the preferred RAID implementation for FreeBSD. However, two additional implementations are available.

At the disk driver level, FreeBSD’s geom system can combine disks into RAID arrays with support for RAID 0, RAID 1, and RAID 3. (RAID 3 is similar to RAID 5 but uses a dedicated parity disk instead of distributing parity among all disks in a pool.) You can stack geoms, so RAID 1+0 and RAID 0+1 are possible as well.

FreeBSD also includes support for RAID 0, RAID 1, and RAID 5 in its logical volume manager, GVinum. However, with the advent of full support for ZFS on FreeBSD, the future of GVinum appears to be in question. It is not yet officially deprecated but no longer seems to be actively maintained.
Disk failure recovery

The Google disk failure study cited on this page should be pretty convincing evidence of the need for some form of storage redundancy in most production environments. At an 8% annual failure rate, your organization needs only 150 hard disks in service to expect an average of one disk failure per month.

JBOD and RAID 0 modes are of no help when hardware problems occur; you must recover your data manually from backups. Other forms of RAID enter a degraded mode in which the offending devices are marked as faulty. The RAID arrays continue to function normally from the perspective of storage clients, although perhaps at reduced performance.

Bad disks must be swapped out for new ones as soon as possible to restore redundancy to the array. A RAID 5 array or two-disk RAID 1 array can tolerate the failure of only a single device. Once that failure has occurred, the array is vulnerable to a second failure.

The specifics of the process are usually pretty simple. You replace the failed disk with another of similar or greater size, then tell the RAID implementation to replace the old disk with the new one. What follows is an extended period during which the parity or mirror information is rewritten to the new, blank disk. This is typically an overnight operation. The array remains available to clients during this phase, but performance is likely to be poor.

To limit downtime and the vulnerability of the array to a second failure, most RAID implementations let you designate one or more disks as “hot” spares. When a failure occurs, the faulted disk is automatically swapped for a spare, and the process of resynchronizing the array begins immediately. Where supported, hot spares should be used as a matter of course.
**Drawbacks of RAID 5**

RAID 5 is a popular configuration, but it has some weaknesses, too. The following issues apply to RAID 6 also, but for simplicity we frame the discussion in terms of RAID 5.

First, it’s critically important to note that RAID 5 does not replace regular off-line backups. It protects the system against the failure of one disk—that’s it. It does not protect against the accidental deletion of files. It does not protect against controller failures, fires, hackers, or any number of other hazards.

Second, RAID 5 isn’t known for its great write performance. RAID 5 writes data blocks to N–1 disks and parity blocks to the N\textsuperscript{th} disk. Parity data is distributed among all the drives in the array; each stripe has its parity stored on a different drive. Since there’s no dedicated parity disk, it’s unlikely that any single disk will act as a bottleneck.

Whenever a random block is written, at least one data block and the parity block for that stripe must be updated. Furthermore, the RAID system doesn’t know what the new parity block ought to contain until it has read the old parity block and the old data. Each random write therefore expands into four operations: two reads and two writes. (Sequential writes may fare better if the implementation is smart.)

Finally, RAID 5 is vulnerable to corruption in certain circumstances. Its incremental updating of parity data is more efficient than reading the entire stripe and recalculating the stripe’s parity from the original data. On the other hand, it means that at no point is parity data ever validated or recalculated. If any block in a stripe should fall out of sync with the parity block, that fact will never become evident in normal use; reads of the data blocks will still return the correct data. Only when a disk fails does the problem become apparent. The parity block will likely have been rewritten many times since the occurrence of the original desynchronization. Therefore, the reconstructed data block on the replacement disk will consist of essentially random data.

This kind of desynchronization between data and parity blocks isn’t all that unlikely, either. Disk drives are not transactional devices. Without an additional layer of safeguards, there is no simple way to guarantee that either two blocks or zero blocks on two different disks will be properly updated. It’s quite possible for a crash, power failure, or communication problem at the wrong moment to create data/parity skew.

This problem is known as the RAID 5 “write hole,” and it has received increasing attention over the last ten years or so. The implementors of the ZFS filesystem claim that because ZFS uses variable-width stripes, it is immune to the RAID 5 write hole. That’s also why ZFS calls its RAID implementation RAID-Z instead of RAID 5, though in practice the concept is similar.

Another potential solution is “scrubbing,” validating parity blocks one by one while the array is relatively idle. Most RAID implementations include some form of scrubbing function. You just have to remember to activate it regularly (by initiating it from **cron** or a **systemd** timer).
mdadm: Linux software RAID

The standard software RAID implementation for Linux is called md, the “multiple disks” driver. It’s front-ended by the mdadm command. md supports all the RAID configurations listed above as well as RAID 4. An earlier system known as raidtools is no longer used.

You can also implement RAID on Linux through the logical volume manager (LVM2) or through Btrfs or another filesystem with built-in volume management and RAID features. We address LVM2 starting on this page and next-generation filesystems starting on this page. Generally, these multiple implementations represent different epochs of software development, with mdadm being the earliest and ZFS/Btrfs the most recent.

All these systems are actively maintained, so choose whichever you prefer. Sites without an installed base are best off jumping directly to an all-in-one system like Btrfs.

Creating an array

The following scenario configures a RAID 5 array composed of three identical 1TB hard disks. Although md can use raw disks as components, we prefer to give every disk a partition table for consistency, so we start by running gpard, creating a GPT partition table on each disk and assigning all the disk’s space to a single partition of type “Linux RAID.” It’s not strictly necessary to set the partition type, but it’s a useful reminder to anyone who might inspect the table later.

The following command builds a RAID 5 array from three whole-disk partitions:

```
$ sudo mdadm --create /dev/md/extra --level=5 --raid-devices=3
   /dev/sdf1 /dev/sdg1 /dev/sdh1
mdadm: Defaulting to version 1.2 metadata
mdadm: array /dev/md/extra started.
```

The virtual file /proc/mdstat always contains a summary of md’s status and the status of all the system’s RAID arrays. It is especially useful to keep an eye on the /proc/mdstat file after adding a new disk or replacing a faulty drive. (watch cat /proc/mdstat is a handy idiom.)

```
$ cat /proc/mdstat
Personalities : [linear] [multipath] [raid0] [raid1] [raid6] [raid5]
   [raid4] [raid10]
   2096886784 blocks super 1.2 level 5, 512k chunk, algo 2 [3/2] [UU_]
   >.................] recovery = 0.0% (945840/1048443932)
   finish=535.2min speed=32615K/sec
   bitmap: 0/8 pages [0KB], 65536KB chunk

unused devices: <none>
```
The `md` system does not keep track of which blocks in an array have been used, so it must manually synchronize all the parity blocks with their corresponding data blocks. `md` calls the operation a “recovery” since it’s essentially the same procedure used when you swap out a bad hard disk. It can take many hours on a large array.

Some helpful notifications appear in the system logs, too (usually `/var/log/messages` or `/var/log/syslog`):

```
kernel: md: bind<sd1>
kernnel: md: bind<sdg1>
kernnel: md: bind<sdh1>
kernnel: md/raid:md127: device sdg1 operational as raid disk 1
kernel: md/raid:md127: device sdfs operational as raid disk 0
kernel: md/raid:md127: allocated 3316kB
kernel: md/raid:md127: raid level 5 active with 2 out of 3 devices,
                       algorithm 2
kernel: RAID conf printout:
kernnel: --- level:5 rd:3 wd:2
kernel: disk 0, 0:1, dev:sdfs
kernel: disk 1, 0:1, dev:sdg1
kernel: created bitmap (8 pages) for device md127
mdadm[1174]: NewArray event detected on md device /dev/md127
mdadm[1174]: DegradedArray event detected on md device /dev/md127
kernel: md127: bitmap initialized from disk: read 1 pages, set 15998
       of 15998 bits
kernel: md127: detected capacity change from 0 to 2147212066816
kernel: RAID conf printout:
kernnel: --- level:5 rd:3 wd:2
kernel: disk 0, 0:1, dev:sdfs
kernel: disk 1, 0:1, dev:sdg1
kernel: disk 2, 0:1, dev:sdh1
kernel: md: recovery of RAID array md127
kernel: md: minimum _guaranteed_ speed: 1000 KB/sec/disk.
kernnel: md: using maximum available idle IO bandwidth (but not more
       than 2000000 KB/sec) for recovery.
kernel: md: using 128k window, over a total of 1048443392KB.
mdadm[1174]: RebuildStarted event detected on md device /dev/md127
```

The initial creation command also serves to “activate” the array (make it available for use). On subsequent reboots, most distributions (including all our examples) automatically discover and activate any existing arrays.

Note that you specify a device pathname for the composite array when you run `mdadm --create`. Old-style `md` device paths looked like `/dev/md0`, but when you specify a path under the `/dev/md` directory, as was done in this example, `mdadm` writes your chosen name into the array’s superblock. This measure ensures that you can always locate the array by its logical path, even when the array is autostarted and might be assigned a different array number. As you can see
from the log entries above, the array also has a traditional name (here, /dev/rd127).
/dev/md/extra is just a symbolic link to the actual array device.

**mdadm.conf**: document array configuration

**mdadm** does not technically require a configuration file, but it will use a configuration file if you supply one, typically /etc/mdadm/mdadm.conf or /etc/mdadm.conf. We recommend that you add **ARRAY** entries to the configuration file as you create new arrays. Doing so documents the RAID configuration in a standard place and gives administrators an obvious place to look for information when problems occur.

**mdadm --detail --scan** dumps the current RAID setup in the format required for inclusion in mdadm.conf. For example,

```
$ sudo mdadm --detail --scan
ARRAY /dev/md/extra metadata=1.2 name=ubuntu:extra UUID=b72de2fb:60b3
   03af:3c176048:dc5b6c8b
```

With the addition of this line, **mdadm** can now read **mdadm.conf** at startup or shutdown to easily manage the array. For example, to take down the array created above, we could run

```
$ sudo mdadm -S /dev/md/extra
```

And to start it up again, run

```
$ sudo mdadm -As /dev/md/extra
```

The first of these commands would work even without the **mdadm.conf** file, but the second would not.

We formerly recommended that you add **DEVICE** entries for the components of each array to mdadm.conf, too. We take that back. Device names are more ephemeral these days and **mdadm** is better at finding and identifying array components than it used to be. We don’t think **DEVICE** entries are a best practice anymore.

**mdadm** has a **--monitor** mode in which it runs continuously as a daemon process and raises an alarm when problems are detected on a RAID array. Use this feature! To set it up, add a **MAILADDR** or **PROGRAM** line to your **mdadm.conf** file. A **MAILADDR** notifies you of issues by email, and a **PROGRAM** configuration runs an external reporting tool that you supply (as is useful for integrating with monitoring systems; see Chapter 28).

You also need to arrange for the monitor daemon to run at boot time. All our example distributions have an **init** script that does this for you, but the names and procedures for enabling it are slightly different.
Simulating a failure

What happens when a disk actually fails? Let’s find out! `mdadm` offers the handy option to simulate a failed disk.

```
$ sudo mdadm /dev/md/extra -f /dev/sg1
mdadm: set /dev/sg1 faulty in /dev/md/extra
```

```
$ sudo tail -1 /var/log/messages
```

```
$ cat /proc/mdstat
Personalities : [linear] [multipath] [raid0] [raid1] [raid6] [raid5] [raid4] [raid10]
md127: active raid5 sdh1[3] sdf1[0] sg1[1](F)
   2096886784 blocks super 1.2 level 5, 512k chunk, algo 2 [3/2] [UU_]
unused devices: <none>
```

Because RAID 5 is a redundant configuration, the array continues to function in degraded mode, so users will not necessarily be aware of the problem.

To remove the drive from the RAID configuration, use `mdadm -r`:

```
$ sudo mdadm /dev/md/extra -r /dev/sg1
mdadm: hot removed /dev/sg1 from /dev/md/extra
```

Once the disk has been logically removed, you can shut down the system and replace the drive. Hot-swappable drive hardware lets you make the change without turning off the system or rebooting.

If your RAID components are raw disks, replace them only with an identical drive. You can replace partition-based components with any partition of similar size, which is a good reason to build your arrays on top of partitions rather than raw disks. Still, for bandwidth matching it’s best if the underlying drive hardware is similar. (Of course, if your RAID configuration is built on top of partitions, you must run a partitioning utility to define the partitions appropriately before adding the replacement disk to the array.)

In our example, the failure is just simulated, so we can add the drive back to the array without replacing any hardware:
md immediately starts to rebuild the array. As always, you can see its progress in /proc/mdstat. A rebuild can take hours, so consider this fact in your disaster recovery (and testing!) plans.

```
$ sudo mdadm /dev/mdextra -a /dev/sdg1
mdadm: hot added /dev/sdc1
```
20.9 FILESYSTEMS

Even after a hard disk has been conceptually divided into partitions or logical volumes, it is still not ready to hold files. All the abstractions and goodies described in Chapter 5, *The Filesystem*, must be implemented in terms of raw disk blocks. The filesystem is the code that implements these, and it needs to add a bit of its own overhead and data.

Early systems bundled the filesystem implementation into the kernel, but it soon became apparent that support for multiple filesystem types was an important design goal. UNIX systems developed a well-defined kernel interface that allowed multiple types of filesystems to be active at once. The filesystem interface also abstracted the underlying hardware, so filesystems see approximately the same interface to storage devices as do other UNIX programs that access the disks through device files in `/dev`.

Support for multiple filesystem types was initially motivated by the need to support NFS and filesystems for removable media. But once the floodgates were opened, the “what if” era began; many different groups started to work on improved filesystems. Some were system-specific, and others (such as ReiserFS) were not tied to any particular OS.

Most systems have settled on one or two filesystems as mainstream defaults. These filesystems are rigorously tested along with the rest of the system before stable releases are issued.

The predominant pattern is for systems to officially support one traditional-style filesystem (UFS, ext4, or XFS) and one next-generation filesystem that includes volume management and RAID features (ZFS or Btrfs). Support for the latter options is usually best on physical hardware; cloud systems can use them for data partitions, but sometimes not for the boot disk.

Although other filesystem implementations are often just a package installation away, add-on filesystems do bring risk and potential instability. Filesystems are foundational, so they need to be 100% stable and reliable under all use scenarios. Filesystem developers work hard to achieve this level of robustness, but the risk can’t be entirely eliminated.

Unless you’re setting up a storage pool or data disk for a specific application, we recommend against straying from your systems’ supported filesystems. That’s what the documentation and administrative tools most likely assume.

The upcoming sections describe the most common filesystems and their management in a bit more detail. We first describe the traditional filesystems UFS, ext4, and XFS, then move on to the next-generation systems ZFS (this page) and Btrfs (this page).
20.10 Traditional Filesystems: UFS, ext4, and XFS

UFS, ext4, and XFS have separate code bases and histories, but over time they’ve become eerily similar to one another from an administrative perspective.

These filesystems exemplify the old school approach in which volume management and RAID are implemented separately from the filesystem itself. The filesystems limit themselves to plain-vanilla file storage on block devices of defined size. Their features are more or less limited to those outlined in Chapter 5.

Older filesystems in this category were subject to subtle corruption if power was interrupted in the middle of a write operation, because then disk blocks could contain inconsistent data structures. The fsck command was used at boot time to check filesystems for this kind of problem and to automatically patch the most common issues.

Modern filesystems include a feature called journaling that averts the possibility of this type of corruption. When a filesystem operation occurs, the required modifications are first written to the journal. Once the journal update is complete, a “commit record” is written to mark the end of the entry. Only then is the normal filesystem modified. If a crash occurs during the update, the filesystem can later replay the journal log to reconstruct a perfectly consistent filesystem.

In most cases, only metadata changes are journaled. The actual data to be stored is written directly to the filesystem. Some filesystems can use the journal for data too, but at a significant performance cost.

Journaling reduces the time needed to perform filesystem consistency checks (see the fsck section on this page) to approximately one second per filesystem. Barring some type of hardware failure, the state of a filesystem can almost instantly be assessed and restored.

The Berkeley Fast File System implemented by McKusick et al. in the 1980s was an early standard that spread to many UNIX systems. With some small adjustments, it eventually became known as the UNIX File System (UFS) and formed the basis of several other filesystem implementations, including Linux’s ext series. UFS remains the default filesystem used by FreeBSD.

The “second extended filesystem,” ext2, was for a long time the mainstream Linux standard. It was designed and implemented primarily by Rémy Card, Theodore Ts’o, and Stephen Tweedie. Although the code for ext2 was written specifically for Linux, it is functionally similar to the Berkeley Fast File System.

Ext3 added journaling, and ext4 is a comparatively modest update that raises a few size limits, increases the performance of certain operations, and allows the use of “extents” (disk block ranges) for storage allocation rather than just individual disk blocks. Ext4 is the default filesystem on Debian and Ubuntu.
XFS was developed by Silicon Graphics, Inc., later known as SGI. It was the default filesystem for IRIX, SGI’s version of UNIX, and was one of the first extent-based filesystems. That made it particularly suitable for sites that processed large media files, as many SGI customers did. XFS is the default filesystem on Red Hat and CentOS.
**Filesystem terminology**

Largely because of their common history, many filesystems share some descriptive terminology. The implementations of the underlying objects have often changed, but the terms are still widely used by administrators as labels for fundamental concepts.

“Inodes” are fixed-length table entries, each of which holds information about one file. The term is probably short for “index nodes,” although its exact etymology is unclear. Inodes were originally preallocated at the time a filesystem was created, but some filesystems now create them dynamically as they are needed. Either way, an inode usually has an identifying number, which you can see with `ls -i`.

Inodes are the “things” pointed to by directory entries. When you create a hard link to an existing file, you create a new directory entry, but you do not create a new inode.

A superblock is a record that describes the characteristics of the filesystem. It contains information about the length of a disk block, the size and location of the inode tables, the disk block map and usage information, the size of the block groups, and a few other important parameters of the filesystem. Because damage to the superblock could erase crucial information, several copies of it are maintained in scattered locations.

The kernel caches disk blocks to increase efficiency. All types of blocks can be cached, including superblocks, inode blocks, and directory information. Caches are normally not “write through,” so there might be some delay between the point at which an application thinks it has written a block and the point at which the block is actually saved to disk. Applications can request more predictable behavior for a file, but this option lowers throughput.

The `sync` system call flushes modified blocks to their permanent homes on disk, possibly making the on-disk filesystem fully consistent for a split second. This periodic save minimizes the amount of data loss that might occur if the machine were to crash with many unsaved blocks. Filesystems can do syncs on their own schedule or leave this up to the OS. Modern filesystems have journaling mechanisms that minimize or eliminate the possibility of structural corruption caused by a crash, so sync frequency now mostly has to do with how many data blocks might be lost in a crash.

A filesystem’s disk block map is a table of the free blocks it contains. When new files are written, this map is examined to devise an efficient layout scheme. The block usage summary records basic information about the blocks that are already in use.
**Filesystem polymorphism**

Filesystems are software packages with multiple components. One part lives in the kernel (or even potentially in user space under Linux; search for “FUSE”) and implements the nuts and bolts of translating the standard filesystem API into reads and writes of disk blocks. Other parts are user-level commands that initialize new volumes to the standard format, check filesystems for corruption, and perform other format-specific tasks.

Long ago, the standard user-level commands knew about “the filesystem” that the system used, and they simply implemented the appropriate functionality. `mkfs` or `newfs` created new filesystems, `fsck` fixed problems, and `mount` mostly just invoked the appropriate underlying system calls.

These days, many more filesystems exist, so systems have had to decide how to address this cornucopia of options. For a long time, Linux tried to fit all filesystems into the standard mold of `mkfs` and `fsck` by making those commands be wrappers. The wrappers called discrete commands named, e.g., `mkfs./fsname` or `fsck./fsname` depending on the type of filesystem being manipulated. These days, the pretense of homogeneity among filesystems has been stretched past the breaking point, and most systems now advise you to call the filesystem-specific commands directly.
**Filesystem formatting**

The general recipe for creating a new Linux filesystem is

```
mkfs.fstype [ -L label ] [ other_options ] device
```

On FreeBSD, the process for creating a UFS filesystem is similar, but with `newfs`:

```
newfs [ -L label ] [ other_options ] device
```

The `-L` option to both `mkfs` and `newfs` sets a volume label for the filesystem such as “spare,” “home,” or “extra.” This is just one option among many, but it’s an option that we recommend you use on every filesystem. Labeling the filesystem frees you from having to track the device on which it’s been installed. It’s particularly handy given that disk device names can change whenever hardware is adjusted.

The available `other_options` are filesystem-specific, but their use is uncommon.
fsck: check and repair filesystems

Because of block buffering and the fact that disk drives are not really transactional devices, filesystem data structures can potentially become self-inconsistent. If these problems are not corrected quickly, they propagate and snowball.

The original fix for corruption was a command called `fsck` ("filesystem consistency check," spelled aloud or pronounced "FS check" or "fisk") that carefully inspected all data structures and walked the allocation tree for every file. It relied on a set of heuristic rules about what the filesystem state might look like after failures at various points during an update.

The original `fsck` scheme worked surprisingly well, but because it involved reading all the data on a disk, it could take hours on a large drive. An early optimization was a "filesystem clean" bit that could be set in the superblock when the filesystem was properly unmounted. When the system restarted, it would see the clean bit and know to skip the `fsck` check.

Now, filesystem journals let `fsck` pinpoint the activity that was occurring at the time of a failure. `fsck` can simply rewind the filesystem to the last known consistent state.

Disks are normally `fsck`ed automatically at boot time if they are listed in the system’s `/etc/fstab` file. The `fstab` file has legacy “`fsck` sequence” fields that ordered and parallelized filesystem checks. But now that `fscks` are fast, the only thing that matters is that the root filesystem be checked first.

You can run `fsck` by hand to perform an in-depth examination more akin to the original `fsck` procedure, but be aware of the time required.

Linux ext-family filesystems can be set to force a recheck after they have been remounted a certain number of times or after a certain period of time, even if all the unmounts were “clean.” This precaution is probably good hygiene, and in most cases the default value (usually around 20 mounts) is acceptable. However, on systems that mount filesystems frequently, such as desktop workstations, even that frequency of `fscks` can become tiresome. To increase the interval to 50 mounts, use the `tune2fs` command:

```
$ sudo tune2fs -c 50 /dev/sda3
tune2fs 1.43.3 (04-Sep-2016)
Setting maximal mount count to 50
```

If a filesystem appears damaged and `fsck` cannot automatically repair it, do not experiment with it before making an ironclad backup. The best insurance policy is to `dd` the entire disk to a backup file or backup disk.

Most filesystems create a `lost+found` directory at the root of each filesystem in which `fsck` can deposit files whose parent directory cannot be determined. The `lost+found` directory has some extra space preallocated so that `fsck` can store orphaned files there without having to allocate
additional directory entries on an unstable filesystem. Don’t delete this directory. (Some systems have a **mklost+found** command you can use to re-create this directory if it is deleted.)

Since the name given to a file is recorded only in the file’s parent directory, names for orphan files are not available and so the files placed in **lost+found** are named with their inode numbers. The inode table does record the UID of the file’s owner, however, so getting a file back to its original owner is relatively easy.
**Filesystem mounting**

A filesystem must be mounted before it becomes visible to processes. The mount point for a filesystem can be any directory, but the files and subdirectories beneath it are not accessible while a filesystem is mounted there. See [Filesystem mounting and unmounting](#) for more information.

After installing a new disk, mount new filesystems by hand to be sure that everything is working correctly. For example, the command

```
$ sudo mount /dev/sda1 /mnt/temp
```

mounts the filesystem in the partition represented by the device file `/dev/sda1` (device names will vary among systems) on a subdirectory of `/mnt`, which is a traditional path used to contain temporary mounts.

You can verify the size of a filesystem with the `df` command. The example below uses the Linux `-h` flag to request “human readable” output. Unfortunately, most systems’ `df` defaults to an unhelpful unit such as “disk blocks,” but there is usually a flag to make `df` report something specific such as kibibytes or gibibytes.

```
$ df -h /mnt/web1
Filesystem         Size  Used  Avail  Use% Mounted on
/devmapper/Demo-web1 197G   60M  187G   1%   /mnt/web1
```
Setup for automatic mounting

You will generally want to configure the system to mount local filesystems at boot time. The /etc/fstab file lists the device names and mount points of all the system’s disks (among other things).

`mount`, `umount`, `swapon`, and `fsck` all read the `fstab` file, so it’s helpful if the data presented there is correct and complete. `mount` and `umount` use the catalog to figure out what you want done if you specify only a partition name or mount point on the command line. For example, with the Linux `fstab` configuration shown on this page, the command

```
$ sudo mount /media/cdrom0
```

would have the same effect as typing

```
$ sudo mount -t udf -o user,noauto,exec,utf8 /dev/scd0 /media/cdrom0
```

The command `mount -a` mounts all regular filesystems listed in the filesystem catalog; it is usually executed from the startup scripts at boot time. (The `noauto` mount option excludes a given filesystem from automatic mounting by `mount -a`.)

The `-t fstype` argument constrains the operation to filesystems of a certain type. For example,

```
$ sudo mount -t ext4
```

mounts all local ext4 filesystems. The `mount` command reads `fstab` sequentially. Therefore, filesystems that are mounted beneath other filesystems must follow their parent partitions in the `fstab` file. For example, the line for `/var/log` must follow the line for `/var` if `/var` is a separate filesystem.

The `umount` command for unmounting filesystems accepts a similar syntax. You cannot unmount a filesystem that a process is using as its current directory or on which files are open. Several commands can identify the processes that are interfering with your `umount` attempt; see this page.

The FreeBSD `fstab` file is the most traditional of our example systems. Here’s a sample from a system with only one real filesystem beyond the root (/spare):

```
<table>
<thead>
<tr>
<th>#</th>
<th>Device</th>
<th>Mountpoint</th>
<th>FType</th>
<th>Options</th>
<th>Dump</th>
<th>Pass#</th>
</tr>
</thead>
<tbody>
<tr>
<td>0</td>
<td>/dev/gpt/rootfs</td>
<td>/</td>
<td>ufs</td>
<td>rw</td>
<td>1</td>
<td>1</td>
</tr>
<tr>
<td>1</td>
<td>/dev/gpt/swap-a</td>
<td>none</td>
<td>swap</td>
<td>sw</td>
<td>0</td>
<td>0</td>
</tr>
<tr>
<td>2</td>
<td>/dev/gpt/swap-b</td>
<td>none</td>
<td>swap</td>
<td>sw</td>
<td>0</td>
<td>0</td>
</tr>
<tr>
<td>3</td>
<td>fdesc</td>
<td>/dev/fd</td>
<td>fdescfs</td>
<td>rw</td>
<td>0</td>
<td>0</td>
</tr>
<tr>
<td>4</td>
<td>proc</td>
<td>/proc</td>
<td>procfs</td>
<td>rw</td>
<td>0</td>
<td>0</td>
</tr>
<tr>
<td>5</td>
<td>/dev/gpt/spare</td>
<td>/spare</td>
<td>ufs</td>
<td>rw</td>
<td>0</td>
<td>0</td>
</tr>
</tbody>
</table>
```
Each line holds six fields separated by whitespace. Each line describes a single filesystem. The fields are traditionally aligned for readability, but alignment is not required.

See Chapter 21 for more information about NFS.

The first field gives the device name. The `fstab` file can include mounts from remote systems, in which case the first field contains an NFS path. The notation `server: /export` denotes the `/export` directory on the machine named `server`.

The second field specifies the mount point, and the third field names the type of filesystem. The exact type name used to identify local filesystems varies among machines.

The fourth field specifies `mount` options to be applied by default. There are many possibilities; see the man page for `mount` for the ones that are common to all filesystem types. Individual filesystems usually introduce options of their own.

The fifth and sixth fields are vestigial. They are supposedly a “dump frequency” column and a column that controls `fsck` parallelism. Neither is important on contemporary systems.

The devices listed for `/dev/fd` and `/proc` are dummy entries. These virtual filesystems are task-specific and don’t require any additional information to be mounted. The other devices are identified by their GPT partition labels, which is a more robust option than using actual device names. To find out the label of an existing partition, run

```
gpart show -l disk
```

to print the partition table of the appropriate disk. To set the label on a partition, use

```
gpart modify -i index -l label disk
```

A cautionary note: partition tables are sometimes referred to as “disk labels.” Make sure when reading documentation that you distinguish between the label of an individual partition and the “label” of the disk itself. Overwriting a disk’s partition table is potentially disastrous.

UFS filesystems also have labels of their own, and these show up beneath the `/dev/ufs` directory. The UFS labels and partition labels are separate, but they can be (and probably should be) set to the same value. In this example, `/dev/ufs/spare` would work just as well as `/dev/gpt/spare`. To find a filesystem’s current label, run

```
tunefs -p device
```

To set the label, run

```
tunefs -L label device.
```

Unmount the filesystem before setting the label.
Below are some additional examples culled from an Ubuntu system’s **fstab**. The general format is the same, but Linux systems use a different way to avoid naming disk devices.

```
# <file system> <mount point> <type> <options> <d> <p>
proc   /proc    proc    defaults    0    0
UUID=a8e3..8f8a    /    ext4    errors=remount-ro    0    1
UUID=13e9..b8d2    none    swap    sw    0    0
/dev/scd0    /media/cdrom0    udf,iso9660    user,noauto,exec,utf8    0    0
```

The first line addresses the **/proc** filesystem, which in fact is presented by a kernel driver and has no actual backing store. As in the FreeBSD example above, the **proc** device listed in the first column is just a placeholder.

The second and third lines use filesystem IDs (UUIDs, which we’ve truncated to make the excerpt more readable) instead of device names to identify volumes. This system is similar to the UFS label system used by FreeBSD, except that the identifiers are long random numbers instead of text strings. Use the **blkid** command to discover the UUID of a particular filesystem.

Filesystems can also have administratively assigned labels; use **e2label** or **xfs_admin** to read or set them. If you want to use labels in **fstab** (which is tidier), just substitute **LABEL=label** for **UUID=long-random-number**.

GPT disk partitions can have UUIDs and labels of their own that are independent of the UUIDs and labels of the filesystems they contain. For use of these options to identify partitions in the **fstab** file, the incantations are **PARTUUID=** and **PARTLABEL=**. However, common practice seems to have converged on the use of filesystem UUIDs.

You can also identify devices with pathnames beneath the **/dev/disk** directory. Subdirectories such as **/dev/disk/by-uuid** and **/dev/disk/by-partuuid** are automatically maintained by udev.
USB drive mounting

USB storage devices come in many flavors: personal “thumb” drives, digital cameras, and large external disks, to name a few. Most of these are supported by UNIX systems as data storage devices.

In the past, special tricks were necessary to manage USB devices. But now that operating systems have embraced dynamic device management as a fundamental requirement, USB drives are just one more type of device that shows up or disappears without warning.

From the perspective of storage management, the issues are two-fold:

- Getting the kernel to recognize a device and to assign a device file to it
- Finding out what assignment has been made

The first step usually happens automatically. Once a device file has been assigned, you can use the normal procedures described in *Disk device files* to find out what it is. For additional information about dynamic device management, see *Chapter 11, Drivers and the Kernel.*
Swapping recommendations

Raw partitions or logical volumes, rather than structured filesystems, are normally used for swap space. Instead of using a filesystem to keep track of the swap area’s contents, the kernel maintains its own simplified mapping from memory blocks to swap space blocks.

On some systems, it’s also possible to swap to a file in a filesystem partition. With older kernels this configuration can be slower than using a dedicated partition, but it’s still handy in a pinch. In any event, logical volume managers eliminate most of the reasons you might want to use a swap file rather than a swap volume.

The more swap space you have, the more virtual memory your processes can allocate. The best virtual memory performance is achieved when the swap area is split among several drives. Of course, the best option of all is to not swap; consider adding RAM if you find yourself needing to optimize swap performance.

The proper amount of swap space to allocate depends on how a machine is used. There is no penalty to overprovisioning except that you lose the extra disk space. We suggest half the amount of RAM as a rule of thumb, but never less than 2GB on a physical server.

If a system will hibernate (personal machines, usually), it needs to be able to save the entire contents of memory to swap in addition to saving all the pages that would be swapped in normal operation. On these machines, increase the swap space recommended above by the amount of RAM.

Cloud and virtualized instances have their own peculiarities with respect to swap space. Paging is always a performance killer, so some sources recommend running without swap space entirely; if you need more memory, you need a larger instance. On the other hand, small instances usually have such meager RAM allotments that they can barely boot without a swap area. The general rule is that it’s fine for instances to have swap space as long as you don’t use it at steady state (or pay extra for it). Whatever approach you decide to take, check your base images to see how they’re set up. Some come with swap preconfigured and some don’t.

Some Amazon EC2 instances come with a local “instance store.” This is essentially a slice of a local hard disk on the machine that runs the hypervisor. The contents of the instance store don’t persist across starts and stops. The store is included in the price of the instance, so you may as well use it for swap space if nothing else.

On Linux systems, you initialize swap areas with mkswap, which takes the device name of the swap volume as an argument. mkswap writes some header information to the swap area. That data includes a UUID, which is why swap partitions count as “filesystems” from the perspective of /etc/fstab and can be identified there by UUID.

You can manually enable swapping to a particular device with swapon device. However, you will generally want to have this function automatically performed at boot time. Just list swap
areas in the regular `fstab` file and give them a filesystem type of `swap`.

To review the system’s active swapping configuration, run `swapon -s` on Linux systems or `swapctl -l` on FreeBSD.
20.11 Next-generation Filesystems: ZFS and Btrfs

Although ZFS and Btrfs are usually referred to as filesystems, they represent vertically integrated approaches to storage management that include the functions of a logical volume manager and a RAID controller. Although the current versions of both systems have a few limitations, most fall into the “not yet implemented” category rather than the “can’t do for architectural reasons” category.
Copy-on-write

Both ZFS and Btrfs avoid overwriting data in place and instead use a scheme known as “copy on write.” To update a block of metadata, for example, the filesystem modifies the in-memory copy and then writes it to a previously vacant disk block. Of course, that data block probably has a parent block that points to it, so the parent is rewritten as well, as is the parent’s parent, and so on back to the topmost level of the filesystem. (In practice, caching and careful design of data structures optimize-out most of these writes, at least in the short term.)

The advantage of this architecture is that the on-disk copy of the filesystem remains perpetually consistent. Before the root block is updated, the filesystem looks exactly as it did the last time the root was updated. A few “empty” blocks have been modified, but nothing points to them, so it makes no difference. The filesystem as a whole moves directly from one consistent state to another.
Error detection

ZFS and Btrfs also take data integrity far more seriously than do traditional filesystems. These systems store checksums for every disk block, and they verify all blocks read to ensure that misreads are detected. On storage pools that include mirroring or parity, bad data is automatically reconstructed from a known-good copy.

Disk drives implement their own layers of error detection and error correction, and although they fail frequently, they’re not supposed to do so without reporting an error back to the host computer. Nevertheless, they sometimes do return bad data without an error indication.

One commonly cited rule of thumb is to expect an instance of silent data corruption for every 75TB of data read. A 2008 study by Bairavasundaram et al. examined service records of more than 1.5 million disk drives in NetApp servers and found that 0.5% of drives showed evidence of silent read errors in each year of service. (Interestingly, one key finding of this study was that enterprise-grade hard disks were an order of magnitude less likely to experience these types of errors.)

These error rates are small, but by all indications they’re staying about the same even as disk capacities and the volumes of data stored on disks expand exponentially. Soon we’ll have hard disks so large that you can’t read the entire contents without a better-than-even chance of encountering a silent error. The extra validation done by ZFS and Btrfs is starting to look really important. (A related issue is the risk of random bit errors in RAM. They are infrequent but they do happen. All production servers should use—and monitor!—ECC memory.)

Parity RAID does not address this issue, at least in normal use. Parity can’t be checked without a reading of the contents of an entire stripe, and it’s inefficient to expand every disk access into a full-stripe read. Scrubbing can help find latent errors, but only if they’re reproducible.
Performance

All the traditional filesystems that remain in common use have similar performance. It’s possible to contrive workloads for which one filesystem or another has an edge, but general-purpose benchmarks rarely show much difference.

Copy-on-write filesystems access storage media somewhat differently from traditional filesystems, and they lack the decades of iterative refinement that have brought the old-guard filesystems to their current state of polish. Usually, the traditional filesystems set the upper bound on filesystem performance.

In many benchmarks, ZFS and Btrfs show performance comparable to traditional filesystems. But at their worst, these filesystems can be about half as fast as the traditional options.

Judging from Linux benchmarks (the only platform on which direct comparison is possible, since Btrfs is Linux-only), Btrfs currently has a slight performance edge over ZFS. However, the results vary widely by access pattern. It is not uncommon for one of these filesystems to perform well on a particular benchmark while the other lags far behind.

The performance picture is complicated by the fact that each of these filesystems has some potential tricks up its sleeve to increase performance. Benchmarks usually don’t take account of these end-arounds. ZFS lets you add caching SSDs to a storage pool; it automatically copies frequently read data to the cache and avoids hitting the hard disks entirely. On Btrfs, you can use `chattr +C` to disable copy-on-write semantics for the data in specific files (usually large or frequently modified ones), thereby skirting some common low-performance scenarios.

For general use as root filesystems and home directory storage, ZFS and Btrfs perform well and offer many useful advantages. They can also work well as data storage for specific server workloads. However, in these latter scenarios, it’s worth taking some time to double-check their behavior in your particular environment.
ZFS was introduced in 2005 as a component of OpenSolaris, and it quickly made its way to Solaris 10 and to various BSD-based distributions. In 2008, it became usable as a root filesystem, and it has been the front-line filesystem of choice for Solaris ever since. UFS remains the default root filesystem on FreeBSD, but ZFS has been an officially supported option since FreeBSD 10.

ZFS is more than just a filesystem, RAID controller, and volume manager wrapped into one. As originally conceived for OpenSolaris, it was a comprehensive rethinking of storage-related administration that addressed everything from the way filesystems were mounted to the way they were exported to other systems over NFS and SMB.

Modern BSD and Linux systems need to accommodate a variety of filesystems, so they’ve been forced to back off a bit from ZFS’s original comprehensive approach. Nevertheless, ZFS remains a thoughtfully designed system that solves quite a few administrative problems through its architecture rather than through the addition of features.
ZFS on Linux

Although ZFS is free software, its use on Linux has been hampered by the fact that the source code is covered by Sun Microsystems’ Common Development and Distribution License (CDDL). The Free Software Foundation maintains that the CDDL is incompatible with the GNU Public License, which covers the Linux kernel. Although add-on versions of ZFS for Linux have long been available through the OpenZFS project (openzfs.org), the FSF’s position has discouraged Linux distributions from bundling ZFS into their base systems.

After nearly a decade of impasse over this issue, the FSF’s position is at last being challenged by Canonical Ltd., developers of Ubuntu. After a legal review, Canonical formally disputed the FSF’s interpretation of the GPL and included ZFS in Ubuntu 16.04 in the form of a loadable kernel module. So far (mid 2017), no lawsuit has resulted. If Canonical remains unpunished, it’s possible that ZFS might become a fully supported root filesystem on Ubuntu and that other distributions might follow suit in supporting it.

If nothing else, the story of ZFS is an interesting case in which the GPL has actively impeded the development of an open source software package and blocked its adoption by users and distributors. If you’re interested in the legal details, Richard Fontana’s wrap-up of open source legal news for 2016 at goo.gl/PC9i3t includes a helpful summary.
ZFS architecture

Exhibit C shows a schematic of the major objects in the ZFS system and their relationship to each other.

Exhibit C: ZFS architecture

A ZFS “storage pool” is analogous to a “volume group” in other logical volume management systems. Each pool is composed of “virtual devices,” which can be raw storage devices (disks, partitions, SAN devices, etc.), mirror groups, or RAID arrays. ZFS RAID is similar in spirit to RAID 5 in that it uses one or more parity devices to implement redundancy for the array. However, ZFS calls the scheme RAID-Z and uses variable-sized stripes to eliminate the RAID 5 write hole. All writes to the storage pool are striped across the pool’s virtual devices, so a pool that contains only individual storage devices is effectively an implementation of RAID 0, although the devices in this configuration are not required to be of the same size.

Unfortunately, the current ZFS RAID is a bit brittle: you cannot add new devices to an array once it has been defined; nor can you permanently remove a device. As in most RAID implementations, devices in a RAID set must be the same size. You can force ZFS to accept mixed sizes, but the size of the smallest volume then dictates the overall size of the array. To use disks of different sizes efficiently in combination with ZFS RAID, you must partition the disks ahead of time and define the leftover regions as separate devices.

Most configuration and management of ZFS is done through two commands: `zpool` and `zfs`. Use `zpool` to build and manage storage pools. Use `zfs` to create and manage the entities created from pools, chiefly filesystems and raw volumes used as swap space, database storage, or backing for SAN volumes.
Example: disk addition

Before we descend into the details of ZFS, here’s a high-level example. Suppose you’ve added a new disk to your FreeBSD system and the disk has shown up as `/dev/ada1`. (An easy way to determine the correct device is to run `geom disk list`.)

The first step is to add the disk to a new storage pool:

```bash
$ sudo zpool create demo ada1
```

Step two is… well, there is no step two. ZFS creates the pool “demo,” creates a filesystem root inside that pool, and mounts that filesystem as `/demo`. The filesystem is automatically remounted when the system boots.

```bash
$ ls -a /demo
.
.. 
```

It would be even more impressive if we could simply add our new disk to the existing storage pool of the root disk, which on FreeBSD is called “zroot” by default. (The command would be `sudo zpool add rpool ada1`.) Unfortunately, the root pool can contain only a single virtual device. Other pools can be painlessly extended in this manner, however.
Filesystems and properties

It’s fine for ZFS to automatically create a filesystem on a new storage pool—by default, ZFS filesystems consume no particular amount of space. All filesystems that live in a pool can draw from the pool’s available space.

Unlike traditional filesystems, which are independent of one another, ZFS filesystems are hierarchical and interact with their parent and child filesystems in several ways. You create new filesystems with `zfs create`:

```
$ sudo zfs create demo/new_fs
$ zfs list -r demo
NAME    USED  AVAIL REFER MOUNTPOINT
demo    432K  945G  96K  /demo
demo/new_fs  96K  945G  96K  /demo/new_fs
```

The `-r` flag to `zfs list` makes it recurse through child filesystems. Most other `zfs` subcommands understand `-r`, too. Ever helpful, ZFS automounts the new filesystem as soon as you create it.

To simulate traditional filesystems of fixed size, you can adjust the filesystem’s properties to add a “reservation” (an amount of space reserved in the storage pool for the filesystem’s use) and a quota. This adjustment of filesystem properties is one of the keys to ZFS management, and it’s something of a paradigm shift for administrators who are accustomed to other systems. Here, we set both values to 1GB:

```
$ sudo zfs set reservation=1G demo/new_fs
$ sudo zfs set quota=1G demo/new_fs
$ zfs list -r demo
NAME    USED  AVAIL REFER MOUNTPOINT
demo    1.00G  944G  96K  /demo
demo/new_fs  96K  1024M  96K  /demo/new_fs
```

The new quota is reflected in the AVAIL column for `/demo/new_fs`. Similarly, the reservation shows up immediately in the USED column for `/demo`. That’s because the reservations of `/demo`’s descendant filesystems are included in its size tally. (The REFER column shows the amount of data referenced by the active copy of each filesystem. `/demo` and `/demo/new_fs` have similar REFER values because they’re both empty filesystems, not because there’s any inherent relationship between the numbers.)

Both property changes are purely bookkeeping entries. The only change to the actual storage pool is the update of a block or two to record the new settings. No process goes out to format the 1GB of space reserved for `/demo/new_fs`. Most ZFS operations, including the creation of new storage pools and new filesystems, are similarly lightweight.

Using this hierarchical system of space management, you can easily group several filesystems to guarantee that their collective size does not exceed a certain threshold; you need not specify
limits on individual filesystems.

You must set both the quota and reservation properties to properly emulate a traditional fixed-size filesystem. The reservation alone simply ensures that the filesystem has enough room available to grow \textit{at least} that large. The quota limits the filesystem’s maximum size \textit{without} guaranteeing that space is available for this growth; another object could snatch up all the pool’s free space, leaving no room for /demo/new_fs to expand.

On the other hand, there are few reasons to set up a filesystem this way in real life. We show the use of these properties simply to demonstrate ZFS’s space accounting system and to emphasize that ZFS is compatible with the traditional model, should you wish to enforce it.
Property inheritance

Many properties are naturally inherited by child filesystems. For example, if we wanted to mount the root of the demo pool in `/mnt/demo` instead of `/demo`, we could simply set the root’s mountpoint parameter:

```
$ sudo zfs set mountpoint=/mnt/demo demo
$ zfs list -r demo
```

```
<table>
<thead>
<tr>
<th>NAME</th>
<th>USED</th>
<th>AVAIL</th>
<th>REFER</th>
<th>MOUNTPOINT</th>
</tr>
</thead>
<tbody>
<tr>
<td>demo</td>
<td>1.00G</td>
<td>944G</td>
<td>96K</td>
<td>/mnt/demo</td>
</tr>
<tr>
<td>demo/new_fs</td>
<td>96K</td>
<td>1024M</td>
<td>96K</td>
<td>/mnt/demo/new_fs</td>
</tr>
</tbody>
</table>
```

```
$ ls /mnt/demo
new_fs
```

Setting the `mountpoint` parameter automatically remounts the filesystems, and the mount point change affects child filesystems in a predictable and straightforward way. The usual rules regarding filesystem activity still apply, however; see [this page](#).

Use `zfs get` to see the effective value of a particular property; `zfs get all` dumps them all. The `SOURCE` column tells you why each property has its particular value: `local` means that the property was set explicitly, and a dash means that the property is read-only. If the property value is inherited from an ancestor filesystem, `SOURCE` shows the details of that inheritance as well.

```
$ zfs get all demo/new_fs
```

```
<table>
<thead>
<tr>
<th>NAME</th>
<th>PROPERTY</th>
<th>VALUE</th>
<th>SOURCE</th>
</tr>
</thead>
<tbody>
<tr>
<td>demo/new_fs</td>
<td>type</td>
<td>filesystem</td>
<td>-</td>
</tr>
<tr>
<td>demo/new_fs</td>
<td>creation</td>
<td>Mon Apr 03 0:12 2017</td>
<td>-</td>
</tr>
<tr>
<td>demo/new_fs</td>
<td>used</td>
<td>96K</td>
<td>-</td>
</tr>
<tr>
<td>demo/new_fs</td>
<td>available</td>
<td>1024M</td>
<td>-</td>
</tr>
<tr>
<td>demo/new_fs</td>
<td>referenced</td>
<td>96K</td>
<td>-</td>
</tr>
<tr>
<td>demo/new_fs</td>
<td>compressor</td>
<td>1.00x</td>
<td>-</td>
</tr>
<tr>
<td>demo/new_fs</td>
<td>mounted</td>
<td>yes</td>
<td>-</td>
</tr>
<tr>
<td>demo/new_fs</td>
<td>quota</td>
<td>1G</td>
<td>local</td>
</tr>
<tr>
<td>demo/new_fs</td>
<td>reservation</td>
<td>1G</td>
<td>local</td>
</tr>
<tr>
<td>demo/new_fs</td>
<td>mountpoint</td>
<td>/mnt/new_fs</td>
<td>inherited from demo</td>
</tr>
<tr>
<td>demo/new_fs</td>
<td>checksum</td>
<td>on</td>
<td>default</td>
</tr>
<tr>
<td>demo/new_fs</td>
<td>compression</td>
<td>off</td>
<td>default</td>
</tr>
</tbody>
</table>
... <many more, about 55 in all>

Vigilant readers might notice that the `available` and `referenced` properties look suspiciously similar to the `AVAIL` and `REFER` columns shown by `zfs list`. In fact, `zfs list` is just a different way of displaying filesystem properties. If we had included the full output of our `zfs get` command above, there would be a `used` property in there, too. Use the `-o` option to specify the properties you want `zfs list` to show.
It wouldn’t make sense to assign values to used and to the other size properties, so these properties are read-only. If the specific rules for calculating used don’t meet your needs, other properties such as usedbychildren and usedbysnapshots may give you better insight into how your disk space is being consumed.

You can set additional, nonstandard properties on filesystems for your own use and for the use of your local scripts. The process is the same as that for standard properties. For example, many backup and snapshot utilities for ZFS read their configuration information from filesystem properties.

The names of custom properties must include a colon to distinguish them from standard properties.
One filesystem per user

Since filesystems consume no space and take no time to create, the optimal number of them is closer to “a lot” than “a few.” If you keep users’ home directories on a ZFS storage pool, you may find it helpful to make each home directory a separate filesystem.

There are several benefits:

- If you need to set disk usage quotas, home directories are a natural granularity at which to do this. You can set quotas on both individual users’ filesystems and on the filesystem that contains all users.

- Snapshots are per filesystem. If each user’s home directory is a separate filesystem, the user can access old snapshots through ~/.zfs. This feature alone is a huge timesaver for administrators because it means that users can service most of their own file restore needs. (The .zfs directory is hidden by default. You can make it visible with zfs set snapdir=visible filesystem.)

- ZFS lets you delegate permission to perform various operations such as taking snapshots or rolling back the filesystem to an earlier state. If you prefer, you can give users control over these operations for their own home directories. We do not describe the details of ZFS permission management in this book, however; see the man page entry for zfs allow.
Snapshots and clones

Just like a logical volume manager, ZFS brings copy-on-write to the user level by allowing you to create instantaneous snapshots. However, there’s an important difference: ZFS snapshots are implemented per filesystem rather than per volume, so they have arbitrary granularity.

On the command line, you create snapshots with `zfs snapshot`. For example, the following command sequence illustrates creation of a snapshot, use of the snapshot through the filesystem’s `.zfs/snapshot` directory, and reversion of the filesystem to its previous state.

```bash
$ sudo touch /mnt/demo/new_fs/now_you_see_me
$ ls /mnt/demo/new_fs
now_you_see_me
$ sudo zfs snapshot demo/new_fs@snap1
$ sudo rm /mnt/demo/new_fs/now_you_see_me
$ ls /mnt/demo/new_fs
$ ls /mnt/demo/new_fs/.zfs/snapshot/snap1
now_you_see_me
$ sudo zfs rollback demo/new_fs@snap1
$ ls /opt/demo/new_fs
now_you_see_me
```

You assign a name to each snapshot at the time it’s created. The complete specifier for a snapshot is usually written in the form `filesystem@snapshot`.

Use `zfs snapshot -r` to create snapshots recursively. The effect is the same as executing `zfs snapshot` on each contained object individually: each subcomponent receives its own snapshot. All the snapshots have the same name, but they’re logically distinct because the `filesystem` portion is different.

ZFS snapshots are read-only, and although they can bear properties, they are not true filesystems. However, you can instantiate a snapshot as a full-fledged, writable filesystem by “cloning” it:

```bash
$ sudo zfs clone demo/new_fs@snap1 demo/subclone
$ ls /mnt/demo/subclone
now_you_see_me
$ sudo touch /mnt/demo/subclone/and_me_too
$ ls /mnt/demo/subclone
and_me_too now_you_see_me
```

The snapshot that is the basis of the clone remains undisturbed and read-only. However, the new filesystem (`demo/subclone` in this example) retains a link to both the snapshot and the filesystem on which it’s based, and neither of those entities can be deleted as long as the clone exists.

Cloning isn’t a common operation, but it’s the only way to create a branch in a filesystem’s evolution. The `zfs rollback` operation demonstrated above can only return a filesystem to its most recent snapshot, so to use it you must permanently delete (`zfs destroy`) any snapshots made
since the snapshot that is your reversion target. Cloning lets you go back in time without losing access to recent changes.

For example, suppose that you’ve discovered a security breach that occurred some time within the last week. For safety, you want to return a filesystem to its state of a week ago to be sure today that it contains no hacker-installed back doors. At the same time, you don’t want to lose recent work or the data for forensic analysis. The solution is to clone the week-ago snapshot to a new filesystem, `zfs rename` the old filesystem, and then `zfs rename` the clone in place of the original filesystem.

For good measure, also `zfs promote` the clone; this operation inverts the relationship between the clone and the filesystem of origin. After promotion, the main-line filesystem has access to all the old filesystem’s snapshots, and the old, moved-aside filesystem becomes the “cloned” branch.
**Raw volumes**

You create swap areas and raw storage areas with `zfs create`, just as you create filesystems. The `-V size` argument makes `zfs` treat the new object as a raw volume instead of a filesystem. The `size` can use any common unit, for example, `128m`.

Since the volume does not contain a filesystem, it is not mounted; instead, it shows up in the `/dev/zvol` directory and can be referenced as if it were a hard disk or partition. ZFS mirrors the hierarchical structure of the storage pool in these directories, so `sudo zfs create -V 128m demo/swap` creates a 128MB swap volume located at `/dev/zvol/demo/swap`.

You can create snapshots of raw volumes just as you can with filesystems, but because there’s no filesystem hierarchy in which to put a `.zfs/snapshot` directory, the snapshots show up in the same directory as their source volumes. Clones work too, just as you’d expect.

By default, raw volumes receive a space reservation equal to their specified size. You’re free to reduce the reservation or to do away with it entirely, but note that this configuration can make writes to the volume return an “out of space” error. Clients of raw volumes might not be designed to deal with such an error.
Storage pool management

Now that we’ve waded into some of the features that ZFS offers at the filesystem and block-client level, we can go for a longer swim in ZFS’s storage pools.

Up to now, we’ve used a pool called “demo” that we created from a single disk back on this page. Here it is in the output of `zpool list`:

```
$ zpool list
NAME    SIZE  ALLOC  FREE  EXPANDSZ FRAG   CAP  DEDUP  HEALTH ALTROOT
demo    976M  518K   976G -  0%    0%   1.00x  ONLINE -
zroot   19.9G  16.3G  3.61G -  24%   81%  1.00x  ONLINE -
```

The pool named “zroot” contains the bootable root filesystem. Bootable pools are currently restricted in several ways: they can contain only a single virtual device, and that device must be either a mirror array or a single disk drive; it cannot be a striping set or a RAID-Z array. (This is either an implementation limit or a strong push in the direction of robustness for the root filesystem; we’re not sure which.)

`zpool status` adds more detail about the virtual devices that make up a storage pool and reports their current status.

```
$ zpool status demo
 pool: demo
 state: ONLINE
 scan: none requested
 config:

       NAME  STATE READ  WRITE CKSUM
       demo  ONLINE  0     0     0
       ada1  ONLINE  0     0     0

 errors: No known data errors
```

Time to get rid of this demo pool and set up something a bit more sophisticated. We’ve attached five 1TB drives to our example system. We first create a pool called “monster” that includes three of those drives in a RAID-Z single-parity configuration.

```
$ sudo zpool destroy demo
$ sudo zpool create monster raidz1 ada1 ada2 ada3
$ zfs list monster
 NAME      USED  AVAIL REFER MOUNTPOINT
 monster    87.2K 1.84T 29.3K /monster
```

ZFS also understands `raidz2` and `raidz3` for double and triple parity configurations. The minimum number of disks is always one more than the number of parity devices. Here, one drive out of three is used for parity, so roughly 2TB is available for use by filesystems.
For illustration, we then add the remaining two drives configured as a mirror:

```bash
$ sudo zpool add monster mirror ada4 ada5
invalid vdev specification
use '-f' to override the following errors:
mismatched replication level: pool uses raidz and new vdev is mirror
$ sudo zpool add -f monster mirror ada4 ada5
```

*zpool* initially balks at this configuration because the two virtual devices have different redundancy schemes. This particular configuration is OK since both vdevs have some redundancy. In actual use, do not mix redundant and nonredundant vdevs since there’s no way to predict which blocks might be stored on which devices; partial redundancy is useless.

```bash
$ zpool status monster
pool: monster
  state: ONLINE
    scan: none requested
  config:

    NAME    STATE  READ WRITE CKSUM
    monster ONLINE 0 0 0
    raidz1-0 ONLINE 0 0 0
    ada1   ONLINE 0 0 0
    ada2   ONLINE 0 0 0
    ada3   ONLINE 0 0 0
    mirror-1 ONLINE 0 0 0
    ada4   ONLINE 0 0 0
    ada5   ONLINE 0 0 0

errors: No known data errors
```

ZFS distributes writes among all a storage pool’s virtual devices. As demonstrated in the preceding example, it is not necessary for all virtual devices to be the same size. (In this example the *disks* are all the same size, but the virtual devices are not.) However, the components within a redundancy group should be of similar size. If they are not, only the smallest size is used on each component. Multiple simple disks used together in a storage pool is essentially a RAID 0 configuration.

You can add additional vdevs to a pool at any time. However, existing data won’t be redistributed to take advantage of parallelism. Unfortunately, you cannot currently add additional devices to an existing RAID array or mirror. This is an area in which Btrfs has a distinct advantage, since it accommodates all sorts of reorganizations in a relatively clean and automatic manner.

ZFS has an especially nice implementation of read caching that makes good use of SSDs. To set up this configuration, just add the SSDs to the storage pool as vdevs of type *cache*. The caching system uses an adaptive replacement algorithm developed at IBM that is smarter than a normal
LRU (least recently used) cache. It knows about the frequency at which blocks are referenced as well as their recentness of use, so reads of large files are not supposed to wipe out the cache.

Hot spares are handled as vdevs of type spare. You can add the same disk to multiple storage pools; whichever pool experiences a disk failure first gets to claim the spare disk.
20.13 Btrfs: “ZFS lite” for Linux

Oracle’s Btrfs filesystem project (“B-tree file system,” officially pronounced “butter FS” or “better FS,” though it’s hard not to think “butter face”) aimed to repeat many of ZFS’s advances on the Linux platform during the long interregnum when ZFS seemed like it might be lost to Linux because of licensing issues.

Although Btrfs remains under active development, it’s been a standard part of the Linux kernel trunk since 2009. It’s available and ready to use on nearly all Linux systems, and SUSE Enterprise Linux has even made it a supported option for the root filesystem. Because the code base evolves quickly, it’s probably best to avoid Btrfs on stability-oriented distributions such as Red Hat for now; old versions have known issues.
Btrfs vs. ZFS

Because they share some technical underpinnings, comparisons between Btrfs and ZFS are probably inevitable. However, Btrfs is not a ZFS clone, and it doesn’t seek to reproduce ZFS’s architecture. For example, you mount Btrfs volumes just like those of other filesystems, by running the `mount` command or by listing them in the `/etc/fstab` file.

Although Btrfs volumes and their subvolumes exist in a unified namespace, there’s no hierarchical relationship among them. To make a change to a group of Btrfs subvolumes, you must modify each of them individually. Btrfs commands do not operate recursively, and volume properties are not inheritable. This isn’t an omission so much as a design choice: why load up the filesystem (the developers ask) with features that you can emulate in a shell script?

Btrfs reflects this preference for simplicity in a variety of ways. For example, Btrfs storage pools can include only one group of disks in one particular configuration (e.g., RAID 5), whereas ZFS pools can include multiple disk groups as well as caching disks, intent logs, and hot spares.

As is common in the software arena, debates over the relative merits of ZFS and Btrfs tend to become heated and to focus on stylistic distinctions. However, several important differences between the two systems rise above the level of nitpicking and personal preference.

- Btrfs is the clear winner when it comes to changing your hardware configuration; ZFS didn’t even show up for this fight. You can add or remove disks at any time, or even change RAID type, and Btrfs redistributes existing data accordingly while remaining on-line. In ZFS, such changes are usually impossible without your dumping your data to external media and starting over.

- Even without memory-intensive features (such as deduplication) enabled, ZFS functions best with a generous amount of RAM. 2GB is the recommended minimum. That’s a lot of memory for a virtual server.

- ZFS’s ability to cache frequently read data on separate cache SSDs is a killer feature for many use cases, and one for which Btrfs currently has no answer.

- As of 2017, the Btrfs implementations of parity raid (RAID 5 and 6) are not yet ready for production use. That’s not our opinion; it’s the official word from the developers. This is a significant missing feature.
Setup and storage conversion

In this section we demonstrate a few common Btrfs procedures analogous to those shown for ZFS in previous sections. We first set up Btrfs for use on a set of two 1TB hard disks configured for RAID 1 (mirroring):

$ sudo mkfs.btrfs -L demo -d raid1 /dev/sdb /dev/sdc
Label: demo
UUID: 
Node size: 16384
Sector size: 4096
Filesystem size: 1.91TiB
Block group profiles:
   Data: RAID1 1.00GiB
   Metadata: RAID1 1.00GiB
   System: RAID1 8.00MiB
SSD detected: no
Incompat features: extref, skinny-metadata
Number of devices: 2
Devices:
   ID  SIZE     PATH
   1  978.00GiB /dev/sdb
   2  978.00GiB /dev/sdc

$ sudo mkdir /mnt/demo
$ sudo mount LABEL=demo /mnt/demo

We could name any of the component devices in the mount command line, but it’s simplest to just use the label we assigned to the group, “demo.”

The btrfs filesystem usage command shows how the space on these disks is currently being used:
btrfs subcommands can be abbreviated to any unique prefix. For example, btrfs filesystem usage is also accessible as btrfs fu. We spell out commands for clarity and propriety.

The interesting thing to note in the output above is the small initial allocations into the RAID 1 groups for data, metadata, and system blocks. Most disk space remains in an unallocated pool that has no intrinsic structure. The mirroring we requested isn’t imposed on the disks as a whole, just on the blocks that are actually in use. It’s not a rigid structure so much as a policy to be implemented at the level of block groups.

This distinction is key to understanding how Btrfs can adapt to changing requirements and hardware provisioning. Here’s what happens when we store some files into the new filesystem and then add a third disk:
The new disk, /dev/sdd, has become available to the pool, but the existing block groups are fine as they were, so none of them reference the new disk. Future allocations would automatically take advantage of the new disk. If we like, we can force Btrfs to level the data among all disks:

```sh
$ sudo btrfs balance start --full-balance /mnt/demo
Starting balance without any filters.
Done, had to relocate 5 out of 5 chunks
```

Conversion among RAID levels is also a form of balancing. Now that we have three disks available, we can convert to RAID 5:

```sh
$ sudo btrfs balance start -dconvert=raid5 -mconvert=raid5 /mnt/demo
Done, had to relocate 5 out of 5 chunks
```

If we had glanced at the usage data during the conversion, we’d have seen block groups for both RAID 1 and RAID 5 active simultaneously. Disk removals work similarly: Btrfs incrementally copies all blocks to groups that don’t include the leaving disk, and eventually no data remains there.
Volumes and subvolumes

Snapshots and quotas are filesystem-level entities in Btrfs, so it’s helpful to be able to define portions of the file tree as distinct entities. Btrfs calls these “subvolumes.” A subvolume looks a lot like a regular filesystem directory, and in fact, it remains accessible as a subdirectory of its parent volume, as shown below.

```
$ sudo btrfs subvolume create /mnt/demo/sub
Create subvolume '/mnt/demo/sub'
$ sudo touch /mnt/demo/sub/file_in_a_subvolume
$ ls /mnt/demo/sub
file_in_a_subvolume
```

The subvolume is not automatically mounted; it’s visible here as part of the parent volume. However, you can mount a subvolume independently of its parent with the `subvol` mount option. For example,

```
$ mkdir /sub
$ sudo mount LABEL=demo -o subvol=/sub /sub
$ ls /sub
file_in_a_subvolume
```

There is no way to prevent a subvolume from showing up within its parent volume when the parent is mounted. To create the illusion of multiple, independent, noninteracting volumes, just make them subvolumes of the root and mount each of them separately with the `subvol` option. The root itself is not required to be mounted anywhere. In fact, Btrfs lets you specify a volume other than the root to be the default mount target when no `subvol` is requested; see `btrfs subvolume set-default`.

To see or manipulate the full Btrfs hierarchy under this configuration, just mount the root on a scratch directory with `subvol=/`. It’s fine for volumes to be mounted several times and accessible through multiple paths.
Volume snapshots

Btrfs’s version of volume snapshots works a lot like cp, except that copies are shallow and initially share all their storage with the parent volume:

```
$ sudo btrfs subvolume snapshot /mnt/demo/sub /mnt/demo/sub_snap
Create a snapshot of '/mnt/demo/sub' in '/mnt/demo/sub_snap'
```

Unlike ZFS snapshots, Btrfs snapshots are writable by default. In fact, there is no such thing as a “snapshot” per se in Btrfs; a snapshot is just a volume that happens to share some storage with another volume:

```
$ sudo touch /mnt/demo/sub/another_file
$ ls /mnt/demo/sub
another_file  file_in_a_subvolume
$ ls /mnt/demo/sub_snap
file_in_a_subvolume
```

For an immutable snapshot, just pass the -r option to btrfs subvolume snapshot. Btrfs does not make a fundamental distinction between read-only snapshots and writable copies in the way that ZFS does. (In ZFS, writable copies are “clones.” To create one, first make a read-only snapshot, then create a clone based on that snapshot.)

Btrfs does not enforce any particular naming or location conventions when it comes to defining subvolumes and snapshots, so it’s up to you to decide how these entities should be organized and named. The Btrfs documentation at btrfs.wiki.kernel.org suggests a couple of conventions for your consideration.

Btrfs also has no “rollback” operation that resets a volume to its state as of a particular snapshot. Instead, you can just move the original volume aside and mv or copy a snapshot in its place:

```
$ ls /mnt/demo/sub
another_file  file_in_a_subvolume
$ sudo mv /mnt/demo/sub /mnt/demo/sub.old
$ sudo btrfs subvolume snapshot /mnt/demo/sub_snap /mnt/demo/sub
Create a snapshot of '/mnt/demo/sub_snap' in '/mnt/demo/sub'
$ ls /mnt/demo/sub
file_in_a_subvolume
```

Note that this change confuses direct mounts of the subvolume. They’ll need to be remounted afterward.
Shallow copies

The analogy between Btrfs snapshots and cp is more than just coincidence. You cannot create snapshots—as such—of files or of directories that are not subvolume roots. But interestingly, you can create shallow copies of arbitrary files and directories with cp --reflink, even across subvolume boundaries.

This option activates Btrfs-specific magic inside cp that negotiates directly with the filesystem to arrange for copy-on-write duplication. The semantics are identical to those of a normal cp and also perilously close to those of a snapshot.

Btrfs doesn’t track shallow copies for you as it would with snapshots, and it also doesn’t necessarily guarantee perfect point-in-time consistency for actively modified directory hierarchies. But in other respects, the two operations are markedly similar. One nice feature of shallow copies is that they require no special permissions; any user can take advantage of them.

If you specify the cp option in the form --reflink=auto, cp shallow-copies when it can and behaves normally otherwise. That makes it a tempting target for a ~/.bashrc alias:

```bash
alias cp="cp --reflink=auto"
```
20.14 Data backup strategy

On a good day, your main focus within the storage environment is to ensure that performance remains good and that sufficient free space is available. Unfortunately, not every day is a good day. With the Google Labs study finding that a disk drive has less than a 75% chance of surviving for five years, the deck is stacked against us. Always have systems in place to protect valuable data against catastrophic loss, and be prepared to activate your recovery procedure at short notice.

RAID and other data replication schemes protect against the failure of a single facility or piece of hardware. However, there are many other ways to lose data that these technologies do not address. For example, if you experience a security breach or an infection by ransomware, your data can be altered or corrupted even though the physical layer remains perfectly intact. Automated replication of compromised data to multiple disks or sites only increases the misery. You need immutable, point-in-time backups of critical data that you can revert to as a fallback option.

In past decades, media such as magnetic tapes were a popular storage method for off-line backups. However, the capacity of these media proved unable to keep up with the exponentially growing sizes of hard drives and SSDs. Along with the physical challenges of transporting and storing tapes and of maintaining finicky mechanical tape drives, the capacity issues ultimately relegated tape media to the status of 35mm camera film: it’s still technically on the market, but you have to wonder who’s actually buying the stuff.

Today, most cloud platforms let you capture point-in-time backups in the form of snapshots, usually on an automated schedule. You pay a monthly fee for the storage consumed by each snapshot and can set your own retention policies.

Regardless of the exact technology you use to implement backups, you need a written plan that answers at least the following questions.

Overall strategy:

- What data is to be backed up?
- What system or technology will perform the backups?
- Where will backup data be stored?
- Will backups be encrypted? If so, where will encryption keys be stored?
- How much will it cost to store backups over time?

Timelines:
• How often will backups be performed?
• How often will backups be validated and restore-tested?
• How long will backups be retained?

People:
• Who will have access to backup data?
• Who will have access to the encryption keys that protect backup data?
• Who will be in charge of verifying the execution of backups?
• Who will be in charge of validating and restore-testing backups?

Use and protection:
• How will backup data be accessed or restored in an emergency?
• How will you ensure that neither a hacker nor a bogus process can corrupt, modify, or delete backups? (That is, how will you achieve immutability?)
• How will backup data be protected against being taken hostage by an adversarial cloud provider, vendor, or government?

The best answers to these questions vary by organization, type of data, regulatory environment, technology platform, and budget, just to name a few potential factors.

Take time today to map out a backup plan for your environment or to review your existing backup plan.
20.15 Recommended Reading


The two titles above are the go-to references for modern ZFS. Although they purport to be FreeBSD-specific, most of the material applies to ZFS on Linux as well. The *Advanced ZFS* book is particularly useful in its coverage of topics as varied as jails, permission delegation, caching strategies, and performance analysis.


21 The Network File System

The Network File System protocol, commonly known as NFS, lets you share filesystems among computers. NFS is nearly transparent to users, and no information is lost when an NFS server crashes. Clients can simply wait until the server returns and then continue as if nothing had happened.

NFS was introduced by Sun Microsystems in 1984. It was originally implemented as a surrogate filesystem for diskless clients, but the protocol proved to be well designed and useful as a general file sharing solution. These days, all UNIX vendors and Linux distributions offer some version of NFS. The NFS protocol is an open standard that is documented in RFCs (see RFCs 1094, 1813, and 7530 in particular).
21.1 Meet network file services

The goal of a network file service is to grant shared access to files and directories that are stored on the disks of remote systems. User applications must be able to read and write to these files with the same system calls they use for local files; that files are stored elsewhere on the network should be transparent to applications. If more than one network client or application attempts to modify a file simultaneously, the file sharing service must resolve any conflicts that arise.
The competition

See Chapter 22 for more details on SMB and Samba.

NFS is not the only file sharing system around. The Server Message Block (SMB) protocol underlies the file sharing capabilities built into Windows and macOS. However, UNIX and Linux can also speak SMB by running the Samba add-on package. If you run a hybrid network that includes a variety of different operating systems, you may find that SMB is the path that presents the fewest compatibility hurdles.

NFS is most commonly used in shops where UNIX and Linux are predominant. In those contexts it offers a somewhat more natural fit and a higher degree of integration. But even in these environments, SMB remains a plausible option. It’s uncommon—but not unheard of—for sites consisting exclusively of UNIX and Linux systems to rely on SMB as their primary file sharing protocol.

Sharing files over a network seems like a simple task, but in fact it’s a confoundingly complex problem that abounds with edge cases and subtleties. Many protocol issues have come to light only through bugs encountered in unusual situations. Both NFS and SMB show the scars of battles fought to maintain security, performance, and reliability over decades of development and widespread use. Today’s administrators can be confident that these protocols will not regularly corrupt data or otherwise incur the wrath of irate users, but it has taken a lot of work and experience to get to this point.

Storage area network (SAN) systems are another option for high-performance storage management on a network. SAN servers need not understand filesystems because they serve only disk blocks, unlike NFS and SMB, which operate at the level of filesystems and files rather than raw storage devices. A SAN affords fast read/write access, but it’s unable to manage concurrent access by multiple clients without the help of a clustered filesystem.

For big data projects, several open source distributed filesystems have come into common use. GlusterFS and Ceph implement both POSIX-compliant filesystems and RESTful object storage distributed among a cluster of nodes for fault tolerance. Commercial versions of both systems are sold by Red Hat, which acquired both developers. Both systems are production-ready, highly capable filesystems worthy of consideration for use cases like big data processing and high performance computing.

Cloud-based systems have additional options. Refer to this page.
Issues of state

One decision made when designing a network filesystem is to determine what part of the system will track the files that each client has open, information referred to generically as “state.” A server that records the status of files and clients is said to be stateful; one that does is stateless. Both approaches have been used over the years, and both have benefits and drawbacks.

Stateful servers keep track of all open files across the network. This mode of operation introduces many layers of complexity (more than you might expect) and makes recovery in the event of a crash far more difficult. When the server returns from a hiatus, a negotiation between the client and server must occur to reconcile the last known state of the connection. Statefulness lets clients maintain more control over files and facilitates the robust management of files opened in read/write mode.

On a stateless server, each request is independent of the requests that have preceded it. If either the server or the client crashes, nothing is lost in the process. Under this design, it is painless for servers to crash or reboot, since they do not maintain any context. However, it’s impossible for the server to know which clients have opened a file for writing, so it cannot manage concurrency.
Performance concerns

Network filesystems should present a seamless experience to users. Accessing a file over the network should be no different from accessing a file on a local filesystem. Unfortunately, wide area networks have high latencies, which make operations behave erratically, and low bandwidth, which yields slow performance on large files. Most file service protocols, including NFS, incorporate techniques to minimize performance problems on both local and wide area networks.

Most protocols try to minimize the number of network requests. For example, read-ahead caching preloads portions of a file into a local memory buffer to avoid a delay when a new section of the file is read. A little extra network bandwidth is consumed in an effort to avoid the latency of a full round-trip exchange with the server.

Similarly, some systems cache writes in memory and send updates in batches, reducing the delay incurred when communicating write operations to the server. These types of batch operations are referred to generically as request coalescing.
Security

Any service that grants convenient access to files on a network has great potential to cause security problems. Local filesystems implement complex access control algorithms and safeguard files with granular permissions. On a network, these tasks are greatly complicated by differences in configurations among machines and by vagaries such as race conditions, bugs in file service software, and unresolved edge cases in file sharing protocols.

The rise of directory and centralized authentication services has improved the security of network filesystems. The bottom line is that no client can be trusted to authenticate itself sanely, so a trusted, central system must verify identities and approve access to files. Most file sharing services can be integrated with a variety of different authentication providers.

File sharing protocols do not typically address the issues of privacy and integrity—or at least, they do not do so directly. As with authentication, this responsibility is generally outsourced to another layer such as a Kerberos, SSH, or a VPN tunnel. However, recent versions of SMB have added strong encryption and integrity checking. Many sites that run NFS on a trusted LAN choose to forgo cryptography because an easy and high-performance solution is unavailable.
21.2 The NFS Approach

The newest version of the NFS protocol has been refined to increase platform independence, to improve performance over wide area networks such as the Internet, and to add strong, modular security features. Most implementations also include diagnostic utilities to help debug configuration and performance problems.

NFS is a network protocol, so in theory it could be implemented in user space just like most other network services. However, the traditional approach has been for parts of the NFS implementation (on both server and client sides) to live inside the kernel, mostly to improve performance. This general pattern continues even on Linux, where locking functions and certain system calls have proved difficult to export to user space. Fortunately, the kernel-resident parts of NFS need no configuration and are largely transparent to administrators.

NFS is not an off-the-shelf solution for all file sharing problems. High availability can only be achieved with warm standbys, but NFS has no built-in provisions for synchronizing with backup servers. The sudden disappearance of an NFS server from the network can result in clients holding stale file handles that can be cleaned up only with a reboot. Strong security is possible but is overly complex. Despite these drawbacks, NFS remains the most common choice for UNIX and Linux file sharing on a LAN.
Protocol versions and history

The first public release of the NFS protocol was version 2 in 1989. The original protocol made some expensive tradeoffs to favor consistency over performance and was quickly replaced. It’s highly unlikely that you’ll encounter this version in use today.

NFS version 3, which dates from the early 1990s, eliminates this bottleneck with a coherency scheme that permits asynchronous writes. It also updates several other aspects of the protocol that were found to have caused performance problems, and it improves the handling of large files. The net result is that NFS version 3 is quite a bit faster than version 2.

NFS version 4, dating from 2003 but not used widely until later in that decade, is a major overhaul that includes many new fixes and features. Highlighted enhancements include

- Compatibility and cooperation with firewalls and NAT devices
- Integration of the lock and mount protocols into the core NFS protocol
- Stateful operation
- Strong, modular security
- Support for replication and migration
- Support for both UNIX and Windows clients
- Access control lists (ACLs)
- Support for Unicode filenames
- Good performance even on low-bandwidth connections

The various NFS protocol versions cannot talk to one another, but NFS servers (including those on all our example systems) typically implement all three versions. In practice, all combinations of NFS clients and servers can interoperate with some version of the protocol. Always use the V4 protocol if both sides support it.

NFS remains actively developed and in widespread use. Version 4.2, written by some of the original stakeholders from Sun’s heyday, reached RFC draft status in early 2015. The Elastic File System service from AWS, which became generally available in mid-2016, adds NFSv4.1 filesystems for use by EC2 instances.

Although V4 is a significant step forward in many ways, it hasn’t much altered the process of configuring and administering NFS. In some ways this is a feature; for example, you still use the same configuration files and commands to administer all versions of NFS. In other ways it’s a problem; some aspects of the configuration process feel jury-rigged (particularly on FreeBSD),
and some options have become ambiguous or overloaded, with different meanings or configuration formats depending on which version of NFS you are using.
Remote procedure calls

When Sun developed the first versions of NFS in the 1980s, they realized that many of the network-related problems that needed solving for NFS would apply to other network-based services, too. They developed a more general framework for remote procedure calls known as RPC or SunRPC, and built NFS on top of that. This work opened the door for applications of all kinds to run procedures on remote systems as if they were being run locally.

Sun’s RPC system was primitive and somewhat hackish; far better systems exist today to fill this need (as do infinitely more horrifying monstrosities than SunRPC; check out SOAP). Nevertheless, NFS still relies on Sun-style RPCs for much of its functionality. Operations that read and write files, mount filesystems, access file metadata, and check file permissions are all implemented as RPCs. The NFS protocol specification is written generically, so a distinct RPC layer is not technically required. However, we are aware of no NFS implementations that stray from the original architecture in this regard.
**Transport protocols**

NFS version 2 originally used UDP because that was what performed best on the LANs and computers of the 1980s. Although NFS does its own packet sequence reassembly and error checking, UDP and NFS both lack the congestion control algorithms that are essential for good performance on a large IP network.

To remedy these problems (and others), NFS migrated to a choice of UDP or TCP in version 3, and to TCP only in version 4. The TCP option was first explored as a way to help NFS work through routers and over the Internet. Over time, most of the original reasons for preferring UDP over TCP have evaporated in the warm light of fast CPUs, cheap memory, and high-speed networks.
State

A client must explicitly mount an NFS filesystem before using it, just as a client must mount a filesystem stored on a local disk. However, NFS versions 2 and 3 are stateless, and the server does not keep track of which clients have mounted each filesystem. Instead, the server simply discloses a secret “cookie” at the conclusion of a successful mount negotiation. The cookie identifies the mounted directory to the NFS server and so opens a way for the client to access its contents. Cookies persist between reboots of the server, so a crash does not leave the client in an unrecoverable muddle. The client can simply wait until the server is available again and resubmit the request.

NFSv4, on the other hand, is a stateful protocol: both client and server maintain information about open files and locks. When the server fails, the clients assist in the recovery process by sending the server their pre-crash state information. A returning server waits for a predefined grace period for former clients to report their state information before it permits new operations and locks. The cookie management of V2 and V3 no longer exists in NFSv4.
NFS servers maintain a list of directories (called “exports” or “shares”) that they make available to clients over the network. By definition, all servers export at least one directory. Clients can then mount these exports and add them to their `fstab` files.

In V2 and V3, each export is treated as an independent entity that is exported separately. In the V4 specification, a server exports a single hierarchical pseudo-filesystem that incorporates all its exported directories. Essentially, the pseudo-filesystem is the server’s own filesystem namespace skeletonized to remove anything that is not exported.

For example, consider the following list of directories, with the directories to be exported in boldface:

```
/www/domain1
/www/domain2
/www/domain3
/var/logs/httpd
/var/spool
```

In NFS version 3, each exported directory must be separately configured. Client systems must execute three different mount requests to obtain access to all the server’s exports.

In NFS version 4, however, the pseudo-filesystem bridges the disconnected portions of the directory structure to create a single view for NFS clients. Rather than requesting a separate mount for each of `/www/domain1`, `/www/domain2`, and `/var/logs/httpd`, the client can simply mount the server’s entire pseudo-root directory and browse the hierarchy.

The directories that are not exported, `/www/domain3` and `/var/spool`, do not appear during browsing. In addition, individual files contained in `/`, `/var`, `/www`, and `/var/logs` are not visible to the client because the pseudo-filesystem portion of the hierarchy includes only directories. Thus, the client view of the NFSv4-exported filesystem is

```
/
|-- var
    |-- logs
    |    |-- httpd
    `-- www
        |-- domain1
        `-- domain2
```

The server specifies the root of the exported filesystems in a configuration file known as the `exports` file, usually kept in `/etc`. Pure NFSv4 clients cannot peruse the list of mounts on a
remote server. Instead, they simply mount the pseudo-root and then all available exports become accessible through that mount point.

That’s the story according to the RFC specification. In practice, the situation is somewhat fuzzy. The Solaris implementation conformed to this specification. Linux made a halfhearted attempt at support for the pseudo-filesystem in the early NFSv4 code, but later revised it to support the scheme more fully; today’s version appears to respect the intent of the RFC. FreeBSD does not implement the pseudo-filesystem as described by the RFC. The FreeBSD export semantics are essentially the same as in version 3; all subdirectories within an export are available to clients.
**File locking**

File locking (as implemented by the `flock`, `lockf`, or `fcntl` systems calls) has been a sore point on UNIX systems for a long time. On local filesystems, it has been known to work less than perfectly. In the context of NFS, the ground is shakier still. By design, early versions of NFS servers are stateless: they have no idea which machines are using any given file. However, to implement locking, state information is needed. What to do?

The traditional answer was to implement file locking separately from NFS. In most systems, two distinct daemons, `lockd` and `statd`, attempted to make a go of it. Unfortunately, the task was difficult for a variety of subtle reasons, and NFS file locking under `lockd` and `statd` is generally unreliable.

NFSv4 removed the need for `lockd` and `statd` by folding locking (and hence, statefulness and all that it implies) into the core protocol. This change introduces significant complexity but obviates many of the related problems of earlier NFS versions. Unfortunately, separate `lockds` and `statds` are still needed to support V2 and V3 clients if your site has them. Our example systems all ship with the earlier versions of NFS enabled, so the separate daemons still run by default.
**Security concerns**

In many ways, NFS V2 and V3 are poster children for everything that is or ever has been wrong with UNIX and Linux security. The protocol was originally designed with essentially no concern for security, and convenience has its price. NFSv4 has addressed the security concerns of earlier versions by mandating support for strong security services and establishing better means of user identification.

All versions of the NFS protocol are intended to be security-mechanism independent, and most servers support multiple “flavors” of security. A few of the common flavors include:

- AUTH_NONE – no authentication
- AUTH_SYS – UNIX-style user and group access control
- RPCSEC_GSS – a stronger flavor that enables flexible security schemes

Historically, most sites used AUTH_SYS authentication, which depends on UNIX user and group identifiers. In this scheme, the client simply sends the local UID and GID of the user requesting access to the server. The server compares the values to those from its own `/etc/passwd` file (or a network database equivalent such as NIS or LDAP) and determines whether the user should have access. Thus, if users mary and bob share the same UID on two different clients, they will have access to each other’s files. Furthermore, users that have root access on a system can `su` to whatever UID they wish; the server will then give them access to the corresponding user’s files.

Enforcing `passwd` file consistency among systems is essential in environments that use AUTH_SYS. But even this is only a security fig leaf; any rogue host (or heaven forbid, Windows machine) can “authenticate” its users however it likes and thereby subvert NFS security.

See [this page](#) for more information about Kerberos.

To prevent such problems, most sites can use a more robust security mechanism such as Kerberos in combination with the NFS RPCSEC_GSS layer. This configuration requires both the client and server to participate in a Kerberos realm. The Kerberos realm authenticates clients centrally, avoiding the problems of self-identification described above. Kerberos can also provide strong encryption and guaranteed integrity for files transferred over the network. All protocol-conformant NFS version 4 systems must implement RPCSEC_GSS, but it’s optional in version 3.

See [this page](#) for more information about firewalls.
NFS version 4 requires TCP as a transport protocol and communicates over port 2049. Since V4 does not rely on any other ports, opening access through a firewall is as simple as opening TCP port 2049. As with all access list configurations, be sure to specify source and destination addresses in addition to the port. If your site doesn’t need to provide NFS services to hosts on the Internet, block access through the firewall or use a local packet filter.

File service over wide area networks with NFSv2 and V3 is not recommended because of the long history of bugs in the RPC protocols and the lack of strong security mechanisms. Administrators of NFS version 3 servers should block access to TCP and UDP ports 2049 and also the **portmap** port, 111.

Given the myriad and obvious shortcomings of AUTH_SYS security, we strongly recommend discontinuing all use of NFSv3. If you have ancient operating systems that can’t be updated to NFSv4 compatibility, at least implement packet filters to restrict network connectivity.
Identity mapping in version 4

Before launching into the following discussion, we should warn you that we consider all implementations of AUTH_SYS security to be more or less broken for security purposes. We strongly suggest Kerberos and RPCSEC_GSS authentication; it’s the only reasonable choice.

As discussed in Chapter 8, User Management, UNIX operating systems identify users through a collection of UIDs and GIDs in the local passwd file or an LDAP directory. NFS version 4, on the other hand, represents users and groups as string identifiers of the form user@nfs-domain and group@nfs-domain. NFSv4 clients and servers run an identity mapping daemon that translates UNIX identifier values to strings that match this format.

When a V4 client performs operations that return identities, such as listing the owners of a set of files with ls -l (the underlying operation is a series of stat calls), the server’s identity mapping daemon uses its local passwd file to convert the UID and GID of each file object to a string, such as ben@admin.com. The client’s identity mapper then reverses the process, converting ben@admin.com into local UID and GID values, which may or may not be the same as the server’s. If a string value does not match any local identity, the nobody user account is assigned as a placeholder.

At this point, the remote filesystem call (stat) has completed and returned UID and GID values to its caller (here, the ls command). Since ls was called with the -l option, it needs to display text names instead of numbers. So, ls in turn retranslates the IDs back to textual names using the getpwuid and getgrgid library routines. These routines once again consult the passwd file or its network database equivalent. What a long, strange trip it’s been.

Confusingly, the identity mapper is used only when retrieving and setting file attributes, typically ownerships. Identity mapping plays no role in authentication or access control, all of which is handled in the traditional form by RPC. The identity mapper may do a better job of mapping than the underlying NFS protocol, causing the apparent file permissions to conflict with the permissions the NFS server will actually enforce.

Consider, for example, the following commands on an NFSv4 client:

```
[ben@nfs-client]$ id ben
uid=1000(ben) gid=1000(ben) groups=1000(ben)

[ben@nfs-client]$ id john
uid=1010(john) gid=1010(john) groups=1010(john)

[ben@nfs-client]$ ls -ld ben
drwxr-xr-x  2 john root  4096 May 27 16:42 ben

[ben@nfs-client]$ touch ben/file

[ben@nfs-client]$ ls -l ben/file
-rw-r--r--  1 john nfsnobody  0 May 27 17:07 ben/file
```
First, ben is shown to have UID 1000 and john to have UID 1010. An NFS-exported home
directory called **ben** appears to have permissions 755 and is owned by john. However, ben is able
to create a file in the directory even though the `ls -l` output indicates that he lacks write
permission.

On the server, john has UID 1000. Since john has UID 1010 on the client, the identity mapper
performs UID conversion as described above, with the result that “john” appears to be the owner
of the directory. However, the identity mapping daemon plays no role in access control. For the
file creation operation, ben’s UID of 1000 is sent directly to the server, where it is interpreted as
john’s UID and permission is granted.

How do you know which operations are identity mapped and which are not? It’s simple:
whenever a UID or GID appears in the filesystem API (as with `stat` or `chown`), it is mapped.
Whenever the user’s own UIDs or GIDs are used implicitly for access control, they are routed
through the designated authentication system.

For this reason, enforcing consistent `passwd` files or relying on LDAP is essential for users of
AUTH_SYS “security.”

Unfortunately for administrators, identity mapping daemons are not standardized across systems,
so their configuration processes may be different. Specifics for our example systems are covered
[here](#).
Root access and the nobody account

Although users should generally be given identical privileges wherever they go, it’s traditional to prevent root from running rampant on NFS-mounted filesystems. By default, the NFS server intercepts incoming requests made on behalf of UID 0 and changes them to look as if they came from some other user. This modification is called “squashing root.” The root account is not entirely shut out, but it is limited to the abilities of a normal user.

A placeholder account named “nobody” is defined specifically to be the pseudo-user as whom a remote root masquerades on an NFS server. The traditional UID for nobody is 65,534 (the 16-bit twos-complement equivalent of UID -2).

Although the Red Hat NFS server defaults to UID -2, the nobody account in the passwd file uses UID 99. You can leave things as they are, add a passwd entry for UID -2, or change anonuid and anongid to 99 if you wish. It really doesn’t matter. Some systems also have an nfsnobody account.

You can change the default UID and GID mappings for root in the exports file. Some systems have an all_squash option to map all client UIDs to the same pseudo-user UID on the server. This configuration eliminates all distinctions among users and creates a sort of public-access filesystem.

The intent behind these precautions is nice, but their ultimate value is not as great as it might seem. Remember that root on an NFS client can su to whatever UID it wants, so user files are never really protected. The only real effect of root squashing is to prevent access to files that are owned by root and not readable or writable by the world.
Performance considerations in version 4

NFSv4 was designed to achieve good performance over wide area networks. Most WANs have higher latency and lower bandwidth than those of LANs. NFS takes aim at these problems with the following refinements:

- An RPC called COMPOUND clumps multiple file operations into one request, reducing the overhead and latency incurred from multiple remote procedure calls.
- A delegation mechanism allows client-side caching of files. Clients can maintain local control over files, including those open for writing.

These features are part of the core NFS protocol and do not require much attention from system administrators.
21.3 Server-side NFS

An NFS server is said to “export” a directory when it makes the directory available for use by other machines. Exports are presented to NFSv4 clients as a single filesystem hierarchy through the pseudo-filesystem.

In NFS version 3, the process used by clients to mount a filesystem is separate from the process used to access files. The operations use separate protocols, and the requests are served by different daemons: mountd for mount discovery and requests, and nfsd for actual file service. On some systems, these daemons are called rpc.nfsd and rpc.mountd as a reminder that they rely on RPC as an underlying mechanism (and hence require the portmap daemon to be running). In this chapter, we omit the rpc prefix for readability.

NFSv4 does not use mountd at all. If you absolutely must run old clients that only support NFSv3, mountd must remain active.

Both mountd and nfsd should start when the system boots, and both should remain running as long as the system is up. Both Linux and FreeBSD automatically run the daemons when you enable NFS service.

NFS uses a single access-control database that tells which filesystems should be exported and which clients can mount them. The operative copy of this database is usually kept in a file called xtab and also in tables internal to the kernel. xtab is a binary file maintained for use by the server daemon.

Maintaining a binary file by hand is not much fun, so most systems assume that you would rather maintain a text file, usually /etc/exports, that enumerates the system’s exported directories and their access settings. The system can then consult this text file at boot time to automatically construct the xtab file.

/etc/exports is the canonical, human-readable list of exported directories. Its contents are read by exportfs -a on Linux, and at a simple restart of the NFS server on FreeBSD. Hence, when you edit /etc/exports, run exportfs -a to activate your changes on Linux, or run service nfsd restart on FreeBSD. If you serve V3 clients from FreeBSD, restart mountd as well (service mountd reload).

NFS deals with the logical layer of the filesystem. Any directory can be exported; it doesn’t have to be a mount point or the root of a physical filesystem. However, for security, NFS does pay attention to the boundaries between filesystems and does require each device to be exported separately. For example, on a machine that has set up /chimchim/users as a separate partition, you could export /chimchim without implicitly exporting /chimchim/users.

Clients are usually allowed to mount subdirectories of an exported directory if they wish, although the protocol does not require this feature. For example, if a server exports
/chimchim/users, a client could mount only /chimchim/users/joe and ignore the rest of the users directory.
Linux exports

On Linux, the exports file consists of a list of exported directories in the leftmost column followed by the hosts that are allowed to access them and associated options on the right. Whitespace separates the filesystem from the list of clients, and each client is followed immediately by a parenthesized list of comma-separated options. Lines can be continued with a backslash. For example, the line

/home  *.users.admin.com(rw) 172.17.0.0/24(ro)

lets /home be mounted read/write by all machines in the users.admin.com domain, and read-only by all machines on the 172.17.0.0/24 class C network. If a system in the users.admin.com domain resides on the 172.17.0.0/24 network, that client will be granted read-only access. The least privileged rule wins.

Filesystems listed in the exports file without a specific set of hosts are usually mountable by all machines. That’s a sizable security hole.

The same sizable security hole can be created accidentally by a misplaced space. For example, the line

/home  *.users.admin.com (rw)

permits any host read/write access except for *.users.admin.com, which has read-only permission, the default. Oops.

There is unfortunately no way to list multiple client specifications for a single set of options. You must repeat the options for all desired clients. Table 21.1 lists the types of client specifications that can appear in the exports file.

Table 21.1: Client specifications in the Linux /etc/exports file

<table>
<thead>
<tr>
<th>Type</th>
<th>Syntax</th>
<th>Meaning</th>
</tr>
</thead>
<tbody>
<tr>
<td>Hostname</td>
<td>hostname</td>
<td>Individual hosts</td>
</tr>
<tr>
<td>Netgroup</td>
<td>@groupname</td>
<td>NIS netgroups (infrequently used)</td>
</tr>
<tr>
<td>Wild cards</td>
<td>* and ?</td>
<td>FQDNs² with wild cards; * will not match a dot</td>
</tr>
<tr>
<td>IPv4 networks</td>
<td>ipaddr/mask</td>
<td>CIDR-style specifications (e.g., 128.138.92.128/25)</td>
</tr>
<tr>
<td>IPv6 networks</td>
<td>ipaddr/mask</td>
<td>IPv6 addresses with CIDR notation (2001:db8::/32)</td>
</tr>
</tbody>
</table>

² Fully qualified domain names

Table 21.2 shows the most commonly used export options understood by Linux.

Table 21.2: Common export options in Linux
The subtree_check option (the default) verifies that every file accessed by a client lies within an exported subdirectory. If you turn off this option, only the fact that the file is within an exported filesystem is verified. Subtree checking can cause occasional problems when a requested file is renamed while the client has the file open. If you anticipate many such situations, consider setting no_subtree_check.

async tells the NFS server to ignore the protocol spec and reply to requests before they are written to disk. This might result in a slight performance boost, but might also result in corrupted data if the server restarts unexpectedly. The default behavior is sync.

The replicas option is merely a convenience that helps clients discover mirrors if the server goes off-line. The actual replication of the filesystem must be handled out of band through some other mechanism such as rsync or DRBD (replication software for Linux). The replica referral feature was added in NFSv4.1.
Early versions of the Linux NFSv4 implementation required administrators to designate a pseudo-filesystem root with the fsid=0 flag in /etc/exports. This is no longer required. To create a pseudo-filesystem as described by the RFC, just list exports as normal and, from an NFSv4 client, mount / on the server. The subdirectories under the mount point will be the exported filesystems. If you do designate an export as fsid=0, that filesystem and all its subdirectories are exported for V4 clients.
FreeBSD exports

In keeping with longstanding UNIX tradition, the exports format used on FreeBSD is entirely different from that of Linux. Each line in the file (except for lines that start with #, which are comments) is composed of three components: a list of directories to export, the options to apply to those exports, and the set of hosts to which the export applies. As on Linux, a backslash denotes a line continuation.

```
/var/www -ro,alldirs www*.admin.com
```

The line above exports `/var/www` and all of its subdirectories read-only to all hosts matching the pattern `www*.admin.com`. To implement different mount options for different clients, simply repeat the line and specify different values. For example,

```
/var/www -alldirs,sec=krb5p -network 2001:db8::/32
```

allows read/write access for all hosts in the named IPv6 network. Kerberos is used for authentication, integrity, and privacy.

On FreeBSD, exports are per server-filesystem. Multiple exports to the same set of client hosts from the same filesystem must be named on the same line. For example,

```
/var/www1 /var/www2 -ro,alldirs www*.admin.com
```

It would be an error for www1 and www2 to be on separate lines with the same host designations, assuming that www1 and www2 reside within the same filesystem.

To enable NFSv4 you must designate a root by prefixing a line with `v4:`, for example,

```
V4: /exports -sec=krb5p,krb5i,krb5,sys -network *.*.admin.com
```

Only one effective V4 root path is allowed. However, it can be specified more than once with different options for different clients. The root can appear anywhere in the exports file.

The `v4:` line does not actually export any filesystems. It simply chooses a base directory for NFSv4 clients to mount. To activate it, list an export within the root:

```
/exports/www -network *.*.admin.com
```

Despite the V4 root designation, the FreeBSD NFS server does not implement the pseudo-filesystem as described by the RFC. When a V4 root is designated and at least one export is present under that root, a V4 client can mount the root and access all of the files and directories within it regardless of their export status. This information is not clear in the exports(5) documentation, and the ambiguity can be quite dangerous. Do not designate the server’s own filesystem root (`/`) as the V4 root; otherwise, the server’s entire root filesystem will be available to clients.
Because of the V4 root, V2 and V3 clients have a different path to mount than V4 clients have. For example, given the following exports

/exports/www -network 10.0.0.0 -mask 255.255.255.0
V4: /exports -network 10.0.0.0 -mask 255.255.255.0

a V2 or V3 client in the 10.0.0.0/24 network mounts /exports/www, but because of the pseudo-filesystem designation on /exports, a V4 client must mount the export as /www. Alternatively, a V4 client can mount / and access the www directory under that mount point.

Use network ranges for best performance when exporting to a large number of clients. For IPv4 you can use CIDR notation or a subnet mask. For IPv6 you must use CIDR; the -mask option is not permitted. For example:

/var/www -network 10.0.0.0 -mask 255.255.255.0
/var/www -network 10.0.0.0/24
/var/www -network 2001:db8::/32

FreeBSD has fewer export options than Linux affords. Table 21.3 summarizes them.

<table>
<thead>
<tr>
<th>Option</th>
<th>Description</th>
</tr>
</thead>
<tbody>
<tr>
<td>alldirs</td>
<td>Allows mounts at any point in the filesystem</td>
</tr>
<tr>
<td>ro</td>
<td>Exports read-only (read/write is the default)</td>
</tr>
<tr>
<td>o</td>
<td>Synonym for ro; exports read-only</td>
</tr>
<tr>
<td>maproot=xxx</td>
<td>The username or UID to map for access from a remote root user</td>
</tr>
<tr>
<td>mapall=xxx</td>
<td>Maps all client users to the specified user (like maproot)</td>
</tr>
<tr>
<td>sec=flavor</td>
<td>Specifies allowable security methods(^a)</td>
</tr>
</tbody>
</table>

\(^a\) Specify multiple flavors in a comma-separated list, in order of preference. Possible values are sys (UNIX authentication, the default), krb5 (Kerberos authentication), krb5i (Kerberos authentication and integrity), krb5p (Kerberos authentication, integrity, and privacy), and none (anonymous access, not recommended).
nfsd: serve files

Once a client’s mount request has been validated, the client can request various filesystem operations. These requests are handled on the server side by nfsd, the NFS operations daemon. (In reality, nfsd simply makes a nonreturning system call to NFS server code embedded in the kernel.) nfsd does not need to run on an NFS client machine unless the client exports filesystems of its own.

nfsd has no configuration file; options are passed as command-line arguments. You start and stop nfsd with your system’s standard service mechanisms, i.e., systemctl on Linux systems running systemd, and the service command on FreeBSD. Table 21.4 shows which file and option to adjust in order to change the arguments passed to nfsd.

Table 21.4: Where to set startup options for nfsd

<table>
<thead>
<tr>
<th>System</th>
<th>Config file</th>
<th>Option to set</th>
</tr>
</thead>
<tbody>
<tr>
<td>Ubuntu</td>
<td>/etc/default/nfs-kernel-server</td>
<td>RPCNFSDOPTS(^a)</td>
</tr>
<tr>
<td>Red Hat</td>
<td>/etc/sysconfig/nfs</td>
<td>RPCNFSDARGS</td>
</tr>
<tr>
<td>FreeBSD</td>
<td>/etc/rc.conf</td>
<td>nfs_server_flags</td>
</tr>
</tbody>
</table>

\(^a\) Some versions of the nfs-kernel-server package incorrectly suggest that you edit RPCMOUNTDOPTS to set some nfsd options. Do not be fooled.

On Linux systems, run systemctl restart nfs-config.service nfs-server.service to enable nfsd configuration changes. In FreeBSD, use service nfsd restart and service mountd restart.

The -N option to nfsd disables the specified version of NFS. For example, to disable versions 2 and 3, add -N 2 -N 3 to the appropriate file and option specified in Table 21.4 and restart the service. This is a good idea if you are sure you don’t need to support older clients.

nfsd takes a numeric argument that specifies how many server threads to fork. Selecting the appropriate number of nfsds is important and is unfortunately something of a black art. If the number is too low or too high, NFS performance can suffer.

The optimal number of nfsd threads depends on the operating system and the hardware in use. If you notice that ps usually shows the nfsds in state D (uninterruptible sleep) and that some idle CPU is available, consider increasing the number of threads. If you find the load average (as reported by uptime) rising as you add nfsds, you’ve gone too far; back off a bit from that threshold.

Run nfsstat regularly to check for performance problems that might be associated with the number of nfsd threads. See this page for more details on nfsstat.

On FreeBSD, the --minthreads and --maxthreads options to nfsd enable automatic management of the number of threads within the specified bounds. On FreeBSD, see man
rc.conf and refer to the options prefixed with nfs_ for more NFS server settings.
21.4 Client-side NFS

NFS filesystems are mounted in much the same way as local disk filesystems. The `mount` command understands the notation `hostname:directory` to mean the path `directory` on the host `hostname`. As with local filesystems, `mount` maps the remote `directory` on the remote `host` into a directory within the local file tree. After the mount completes, you access an NFS-mounted filesystem just like a local filesystem. The `mount` command and its associated NFS extensions represent the most significant concerns to the system administrator of an NFS client.

Before an NFS filesystem can be mounted, it must be properly exported (see this page). On an NFSv3 client, you can verify that a server has properly exported its filesystems by running the `showmount` command:

```
$ showmount -e monk
Export list for monk:
/home/ben harp.atrust.com
```

This example reports that the directory `/home/ben` on the server monk has been exported to the client system harp.atrust.com.

If an NFS mount is not working, first verify that the filesystems have been properly exported on the server. Make sure that after updating the server’s `exports` file, you ran `exportfs -a` (Linux) or `service nfsd restart` and `service mountd reload` (FreeBSD). Next, recheck the `showmount` output.

If the directory is properly exported on the server but `showmount` returns an error or an empty list, double-check that all the necessary processes are running on the server (`portmap` and `nfsd`; add `mountd`, `statd`, and `lockd` for V3). Make sure the `hosts.allow` and `hosts.deny` files allow access to those daemons and that you are on the right client system.

The path information displayed by `showmount` (e.g., `/home/ben` above) is valid only for NFS version 2 and 3 servers. NFS version 4 servers export a single unified pseudo-filesystem and do not use the mount protocol. The traditional NFS concept of separate mount points doesn’t jibe with version 4’s model, so `showmount` simply doesn’t apply to the V4 world.

Unfortunately, NFSv4 has no good replacement for `showmount`. On the server, the command `exportfs -v` shows the existing exports, but of course this works only locally. If you don’t have direct access to the server, you can try to mount the server’s V4 root and traverse the directory structure manually. You can also mount any subdirectory of the exported root filesystem.

To actually mount the filesystem in versions 2 and 3, you’d use a command such as

```
$ sudo mount -o rw,hard,intr,bg server:/home/ben /nfs/ben
```

To accomplish the same under version 4 on a Linux system, you’d type
In this case, the options after `-o` specify that the filesystem be mounted read/write (`rw`), that operations be interruptible (`intr`), and that retries be done in the background (`bg`). Table 21.5 introduces the most common Linux mount options.

Table 21.5: NFS mount flags and options for Linux

<table>
<thead>
<tr>
<th>Flag</th>
<th>Description</th>
</tr>
</thead>
<tbody>
<tr>
<td><code>rw</code></td>
<td>Mounts the filesystem read/write (must be exported that way)</td>
</tr>
<tr>
<td><code>ro</code></td>
<td>Mounts the filesystem read-only</td>
</tr>
<tr>
<td><code>bg</code></td>
<td>If the mount fails (server doesn’t respond), keeps trying it in the background and continues with other mount requests</td>
</tr>
<tr>
<td><code>hard</code></td>
<td>If a server goes down, makes operations that access it block until the server comes back up</td>
</tr>
<tr>
<td><code>soft</code></td>
<td>If a server goes down, makes operations that access it fail and return an error, to avoid processes hanging on inessential mounts</td>
</tr>
<tr>
<td><code>intr</code></td>
<td>Allows users to interrupt blocked operations (they return an error)</td>
</tr>
<tr>
<td><code>nointr</code></td>
<td>Does not allow user interrupts</td>
</tr>
<tr>
<td><code>retrans=n</code></td>
<td>Specifies the number of times to repeat a request before returning an error on a soft-mounted filesystem</td>
</tr>
<tr>
<td><code>timeo=n</code></td>
<td>Sets the timeout period (in tenths of a second) for requests</td>
</tr>
<tr>
<td><code>rsize=n</code></td>
<td>Sets the read buffer size to <code>n</code> bytes</td>
</tr>
<tr>
<td><code>wsize=n</code></td>
<td>Sets the write buffer size to <code>n</code> bytes</td>
</tr>
<tr>
<td><code>sec=flavor</code></td>
<td>Specifies the security flavor</td>
</tr>
<tr>
<td><code>nfsvers=n</code></td>
<td>Sets the NFS protocol version</td>
</tr>
<tr>
<td><code>proto=proto</code></td>
<td>Selects a transport protocol; must be tcp for NFS version 4</td>
</tr>
</tbody>
</table>

The client side of NFS usually tries to autonegotiate a suitable version of the protocol. You can specify a specific version by passing `-o nfsvers=n`.

On FreeBSD, `mount` is a wrapper that calls `/sbin/mount_nfs` for NFS mounts. This wrapper sets NFS options and invokes the `nmount` system call. To mount a version 4 server on FreeBSD, type:

```
$ sudo mount -t nfs -o nfsvers=4 server:/ /mnt
```

If you don’t specify a version explicitly, `mount` negotiates one automatically in descending order. In fact, a simple `mount server:/ /mnt` does the trick in this case because `mount` can infer from the format that the filesystem you’re referring to is NFS.

Filesystems mounted `hard` (the default) cause processes to hang when their servers go down. This behavior is particularly bothersome when the processes in question are standard daemons, so we
do not recommend serving critical system binaries over NFS. In general, the intr option reduces
the number of NFS-related headaches.

Jeff Forys, one of our technical reviewers, advises, “Most mounts should use hard, intr, and bg,
because these options best preserve NFS’s original design goals. soft is an abomination, an ugly
Satanic hack! If the user wants to interrupt, cool. Otherwise, wait for the server and all will
eventually be well again with no data lost.”

Automount solutions such as autofs, discussed starting here, also prescribe some remedies for
mounting ailments.

The read and write buffer sizes are negotiated to the highest value supported by both client and
server. You can set them to any value between 1KiB and 1MiB.

You can see the available space on an NFS mount with df, just as you would on a local
filesystem:

```
$ df /nfs/ben
Filesystem  1k-blocks  Used  Available  Use% Mounted on
leopard:/home/ben  17212156  1694128  14643692  11% /nfs/ben
```

umount works on NFS filesystems just like it does on local filesystems. If the NFS filesystem is
in use when you try to unmount it, you get an error such as

```
umount: /nfs/ben: device is busy
```

Use fuser or lsof to find processes with open files on the filesystem. Kill them, or in the case of
shells, change directories. If all else fails or your server is down, try running umount -f to force
the filesystem to be unmounted.
Mounting remote filesystems at boot time

See this page for more information about the `fstab` file.

You can use the `mount` command to establish temporary network mounts, but you should list mounts that are part of a system’s permanent configuration in `/etc/fstab` so that they are mounted automatically at boot time. Alternatively, mounts can be handled by an automatic mounting service such as autofs.

The following `fstab` entries mount the `/home` filesystem from the server monk:

```
# filesystem mountpoint fstype flags        dump fsck
monk:/home  /nfs/home  nfs  rw,bg,intr,hard,nodev,nosuid 0   0
```

You can make your changes take effect immediately (without rebooting) by running `mount -a -t nfs`.

The `flags` field of `/etc/fstab` specifies options for NFS mounts; these options are the same ones you would specify on the `mount` command line.
Restricting exports to privileged ports

NFS clients are free to use any TCP or UDP source port they like when connecting to an NFS server. However, some servers may insist that requests come from a privileged port (a port numbered lower than 1,024). Others allow this behavior to be set as an option. The use of privileged ports delivers little actual security.

Nevertheless, most NFS clients adopt the traditional (and still recommended) approach of defaulting to a privileged port to avert the potential for conflict. Under Linux, you can accept mounts from unprivileged ports with the insecure export option.
21.5 Identity Mapping for NFS Version 4

We introduced the general ideas behind NFSv4’s identity mapping system starting here. In this section we discuss the administrative aspects of the identity mapping daemon.

All systems that participate in an NFSv4 network should have the same NFS domain. In most cases, it’s reasonable to use your DNS domain as the NFS domain. For example, admin.com is a straightforward choice of NFS domain for the server ulsal.admin.com. Clients in subdomains (e.g., books.admin.com) may or may not want to use the same domain name (e.g., admin.com) to facilitate NFS communication.

Unfortunately for administrators, NFSv4 UID mapping has no standard implementation, so the details of administration differ slightly among systems. Table 21.6 names the mapping daemons on Linux and FreeBSD and notes the location of their configuration files.

<table>
<thead>
<tr>
<th>System</th>
<th>Daemon</th>
<th>Configuration file</th>
<th>man page</th>
</tr>
</thead>
<tbody>
<tr>
<td>Linux</td>
<td>/usr/sbin/rpc.idmapd</td>
<td>/etc/idmapd.conf</td>
<td>nfsidmap(5)</td>
</tr>
<tr>
<td>FreeBSD</td>
<td>/usr/sbin/nfsuserd</td>
<td>nfsuserd_flags in /etc/rc.conf</td>
<td>idmap(8)</td>
</tr>
</tbody>
</table>

Other than having their NFS domains set, identity mapping services require little assistance from administrators. The daemons are started at boot time by the same scripts that manage other NFS daemons. After making configuration changes, you’ll need to restart the identity mapper daemon. Options such as verbose logging and alternative management of the nobody account are usually available.
21.6 NFSSTAT: DUMP NFS STATISTICS

*nfsstat* displays various statistics maintained by the NFS system. *nfsstat -s* shows server-side statistics, and *nfsstat -c* shows information for client-side operations. By default, *nfsstat* shows statistics for all protocol versions. For example:

```
$ nfsstat -c

Client rpc:
calls  badcalls  retrans  badxid  timeout  wait  newcred  timers
64235  1595       0       3      1592       0      0      886

Client nfs:
calls  badcalls  nclget  nclsleep
62613    3      62643      0
null  getattr  setattr  readdir  lookup  root  read
 0%  34%   0%   21%     30%     0%     2%
write  wrcache  create  remove  rename  link  symlink
 3%  0%    0%    0%     0%     0%     0%
mkdir  readdir  rmdir  fsstat
 0%    6%   0%    0%
```

This example is from a relatively healthy NFS client. If more than 3% of RPC calls time out, it’s likely that your NFS server or network has a problem. You can usually discover the cause by checking the *badxid* field. If *badxid* is near 0 with timeouts greater than 3%, packets to and from the server are getting lost on the network. You might be able to solve this problem by lowering the *rsize* and *wsize* mount parameters (read and write block sizes).

If *badxid* is nearly as high as *timeout*, then the server is responding, but too slowly. Either replace the server or increase the *timeo* mount parameter.

Running *nfsstat* and *netstat* occasionally and becoming familiar with their output helps you discover NFS problems before your users do. We suggest including this data as part of your site’s monitoring and alerting system.
21.7 Dedicated NFS file servers

Fast, reliable file service is an essential element of a production computing environment. Although you can certainly roll your own file servers from workstations and off-the-shelf hard disks, doing so is often not the best-performing or easiest-to-administer solution (though it is usually the cheapest).

Dedicated NFS file server products have been on the market for many years. They offer a host of potential advantages over the homebrew approach:

- They are optimized for file service and typically deliver the best possible NFS performance.

- As storage requirements grow, they can scale smoothly to support tera-bytes of storage and hundreds of users.

- They are more reliable than stand-alone boxes thanks to their simplified software, redundant hardware, and use of disk mirroring.

- They usually handle file service for both UNIX and Windows clients. Most even contain integrated HTTPS, FTP, and SFTP servers.

- They often include backup and checkpoint facilities that are superior to those found on vanilla UNIX systems.

Some of our favorite dedicated NFS servers are made by NetApp. Their products run the gamut from very small to very large, and their pricing is OK. EMC is another player in the high-end server market. They make good products, but be prepared for sticker shock and build up your tolerance for marketing buzzwords.

In an AWS environment, the Elastic File System service is a scalable NFSv4.1 server-as-a-service that exports filesystems to EC2 instances. Each filesystem can support multiple GiB/s throughput, depending upon the size of the filesystem. See aws.amazon.com/efs for more information.
21.8 AUTOMATIC MOUNTING

Mounting filesystems at boot time by listing them in /etc/fstab can cause administrative headaches on large networks. First, it’s tedious to maintain the fstab file on hundreds of machines, even with help from scripts and configuration management systems. Each host may have slightly different needs and so require individual attention. Second, if shared filesystems are mounted from many different hosts, clients become dependent on many different downstream servers. Chaos ensues when one of those servers crashes. Every command that accesses that server’s mount points will hang.

You can moderate these problems with an automounter, a type of daemon that mounts filesystems when they are referenced and unmounts them when they are no longer being used. In addition to deferring mounts until they are actually needed, most automounters can also accept a list of “replicas” (identical backup copies) for a filesystem. These backups let the network continue to function even when a primary server becomes unavailable.

As described by Edward Tomasz Napierała, author of the FreeBSD automounter, this magic requires the cooperation of several related pieces of software:

- autofs, a kernel-resident filesystem driver that watches a filesystem for mount requests, pauses the calling program, and invokes the automounter to mount the target filesystem before returning control to the caller
- **automountd** and **autounmountd**, which read the administrative configuration and actually mount or unmount filesystems
- **automount**, an administrative utility

For the most part, automounters are transparent to users. Instead of mirroring an actual filesystem, the automounter “makes up” a virtual filesystem hierarchy according to the specifications given in its configuration files. When a user references a directory within the automounter’s virtual filesystem, the **automountd** intercepts the reference and mounts the actual filesystem the user is trying to reach. The NFS filesystem is mounted beneath the autofs filesystem in normal UNIX fashion.

The idea of an automounter originally comes from Sun. The Linux version functionally mimics that of Sun, although it is in fact an independent implementation. FreeBSD maintains yet another implementation, having sacrificed a once widely used automounter, **amd**, in the FreeBSD 10.1 release.

The various **automount** implementations understand three different kinds of configuration files, referred to as “maps”: direct maps, indirect maps, and master maps. Direct and indirect maps contain information about the filesystems to be automounted. A master map lists the direct and indirect maps that **automount** should pay attention to. Only one master map can be active at
once; the default master map is kept in `/etc/auto_master` on FreeBSD and in `/etc/auto.master` on Linux.

A direct map can also be managed as an NIS database or in an LDAP directory, but doing so is tricky.

On most systems, **automount** is a stand-alone command that reads its configuration files, sets up any necessary autofs mounts, and exits. Actual references to automounted filesystems are handled (through autofs) by a separate daemon process, **automountd**. This daemon does its work silently and does not need additional configuration.

On Linux systems, the daemon is called **automount** instead of **automountd**, and the setup function is performed by a system startup script (**systemd** for modern distributions). Linux details are given [here](#). In the following discussion, we refer to the setup command as **automount** and the daemon as **automountd**.

If you change the master map or one of the direct maps that it references, you must rerun **automount** to pick up the changes. With the `-v` option, **automount** shows you the adjustments it’s making to its configuration. You can add `-L` to achieve a dry run effect that lets you examine your configuration and debug problems.

**automount** (autounmountd on FreeBSD) accepts a `-t` argument that tells how long (in seconds) an automounted filesystem can remain unused before being unmounted. The default is 300 seconds (10 minutes). Since an NFS mount whose server has crashed can cause programs that touch it to hang, it’s good hygiene to clean up automounts that are no longer in use; don’t raise the timeout too much. (The other side of this issue is the time required to mount a filesystem. System response is faster and smoother if filesystems aren’t being continually remounted.)
**Indirect maps**

Indirect maps automount several filesystems under a common directory. However, the path of the directory is specified in the master map, not in the indirect map itself. For example, an indirect map might look like this:

```
users  harp:/harp/users
devel  -soft harp:/harp/devel
info   -ro harp:/harp/info
```

The first column names the subdirectory in which each automount should be installed, and subsequent items list the mount options and the NFS path of the filesystem. This example (perhaps stored in `/etc/auto.harp`) tells `automount` that it can mount the directories `/harp/users`, `/harp/devel`, and `/harp/info` from the server harp, with `info` being mounted read-only and `devel` being mounted soft.

In this configuration, the paths on harp and the local host are the same. However, this correspondence is not required.
**Direct maps**

Direct maps list filesystems that do not share a common prefix, such as `/usr/src` and `/cs/tools`. A direct map (e.g., `/etc/auto.direct`) that described both of these filesystems to `automount` might look something like this:

```
/usr/src       harp:/usr/src
/cs/tools      -ro monk:/cs/tools
```

Because they do not share a common parent directory, these automounts must each be implemented with a separate autofs mount. This configuration requires more overhead, but it has the added advantage that the mount point and directory structure are always accessible to commands such as `ls`. Running `ls` on a directory full of indirect mounts can be confusing to users because `automount` doesn’t show the subdirectories until their contents have been accessed. (`ls` doesn’t look inside the automounted directories, so it does not cause them to be mounted.)
Master maps

A master map lists the direct and indirect maps that **automount** should pay attention to. For each indirect map, it also specifies the root directory to be used by the mounts defined in the map.

A master map that referenced the direct and indirect maps shown in the previous examples would look something like this:

```
# Directory  Map
/harp     /etc/auto.harp -proto=tcp
/-        /etc/auto.direct
```

The first column is a local directory name for an indirect map or the special token `/` for a direct map. The second column identifies the file in which the map is stored. You can have several maps of each type. When you specify mount options at the end of a line, they set the defaults for all mounts within the map. Linux administrators should always specify the `-fstype=nfs4` mount flag for NFS version 4 servers.

On most systems, the default options set on a master map entry do not blend with the options specified in the direct or indirect map to which it points. If a map entry has its own list of options, the defaults are ignored. Linux merges the two sets, however. If the same option is specified in both places, the map entry’s value overrides the default.
**Executable maps**

If a map file is executable, it’s assumed to be a script or program that dynamically generates automounting information. Instead of reading the map as a text file, the automounter executes it with an argument (the “key”) that indicates which subdirectory a user has attempted to access. The script prints an appropriate map entry; if the specified key is not valid, the script can simply exit without printing anything.

This powerful feature makes up for many of the deficiencies in automounter’s rather strange configuration system. In effect, you can easily define a site-wide automount configuration file in a format of your own choice. You can write a simple script to decode the global configuration on each machine. Some systems come with a handy `/etc/auto.net` executable map that takes a hostname as a key and mounts all exported filesystems on that host.

Since automount scripts run dynamically as needed, it’s unnecessary to distribute the master configuration file after every change or to convert it preemptively to the automounter format; in fact, the global configuration file can have a permanent home on an NFS server.
**Automount visibility**

When you list the contents of an automounted filesystem’s parent directory, the directory appears empty no matter how many filesystems have been automounted there. You cannot browse the automounts in a GUI filesystem browser.

An example:

```bash
$ ls /portal
$ ls /portal/photos
art_class_2010  florissant_1003  rmp03
blizzard2008  frozen_dead_guy_Oct2009  rmp_030806
boston021130  greenville.021129  steamboat2006
```

The **photos** filesystem is alive and well and is automounted under `/portal`. It’s accessible through its full pathname. However, a review of the `/portal` directory does not reveal its existence. If you had mounted this filesystem through the `fstab` file or a manual `mount` command, it would behave like any other directory and would be visible as a member of the parent directory.

One way around the browsing problem is to create a shadow directory that contains symbolic links to automount points. For example, if `/automounts/photos` is a link to `/portal/photos`, you can `ls` the contents of `/automounts` to discover that `photos` is an automounted directory. References to `/automounts/photos` are still routed through the automounter and work correctly.

Unfortunately, these symbolic links require maintenance and can go out of sync with the actual automounts unless they are periodically reconstructed by a script.
Replicated filesystems and automount

In some cases, a read-only filesystem such as `/usr/share` might be identical on several different servers. In this case, you can tell `automount` about several potential sources for the filesystem. It then chooses a server according to its own idea of which servers are closest, given network routes, NFS protocol versions, and response times to an initial query.

Although `automount` itself does not see or care how the filesystems it mounts are used, replicated mounts should represent read-only filesystems such as `/usr/share` or `/usr/local/X11`. There’s no way for `automount` to synchronize writes across a set of servers, so replicated read/write filesystems are of little practical use.

You can assign explicit priorities to determine which replica to select first. The priorities are small integers, with larger numbers indicating lower priority. The default priority is 0, most eligible.

An `auto.direct` file that defines `/usr/man` and `/cs/tools` as replicated filesystems might look like this:

```
/usr/man    -ro harp:/usr/share/man monk(1):/usr/man
/cs/tools   -ro leopard,monk:/cs/tools
```

Note that server names can be listed together if the source path on each is the same. The (1) after monk in the first line sets that server’s priority with respect to `/usr/man`. The lack of a priority after harp indicates an implicit priority of 0.
Automatic automounts (V3; all but Linux)

Instead of listing every possible mount in a direct or indirect map, you can tell automount a little about your filesystem naming conventions and let it figure things out for itself. The key piece of glue that makes this work is that the mountd running on a remote server can be queried to find out what filesystems the server exports. In NFS version 4, the export is always /, which eliminates the need for this automation.

“Automatic automounts” can be configured in several ways, the simplest of which is the -hosts mount type on FreeBSD. If you list -hosts as a map name in your master map file, automount then maps remote hosts’ exports into the specified automount directory:

/net -hosts -nosuid,soft

For example, if harp exported /usr/share/man, that directory could then be reached through the automounter at the path /net/harp/usr/share/man.

The implementation of -hosts does not enumerate all possible hosts from which filesystems can be mounted; that would be impossible. Instead, it waits for individual subdirectory names to be referenced, then runs off and mounts the exported filesystems from the requested host.

A similar but finer-grained effect can be achieved with the * and & wild cards in an indirect map file. Also, a number of macros available for use in maps expand to the current hostname, architecture type, and so on. See the automount(1M) man page for details.
Specifics for Linux

The Linux implementation of automount has diverged a bit from the original Sun standards. The changes mostly relate to the naming of commands and files.

First, automount is the daemon that actually mounts and unmounts remote filesystems. It fills the same niche as the automountd daemon on other systems and generally does not need to be run by hand.

The default master map file is /etc/auto.master. Its format and the format of indirect maps are as described previously. The documentation can be hard to find, however. The master map format is described in auto.master(5) and the indirect map format in autofs(5); be careful, or you’ll get autofs(8), which documents the syntax of the autofs command. (As one of the man pages says, “The documentation leaves a lot to be desired.”) To cause changes to the master map to take effect, run the command /etc/init.d/autofs reload, which is equivalent to automount in Sunland.

The Linux implementation does not support the Solaris-style -hosts clause for automatic automounts.
21.9 **RECOMMENDED READING**

*Table 21.7* lists the various RFCs for the NFS protocol.

**Table 21.7: NFS-related RFCs**

<table>
<thead>
<tr>
<th>RFC</th>
<th>Title</th>
<th>Author</th>
<th>Date</th>
</tr>
</thead>
<tbody>
<tr>
<td>2624</td>
<td>NFS Version 4 Design Considerations</td>
<td>S. Shepler</td>
<td>Jun 1999</td>
</tr>
</tbody>
</table>
Chapter 21, *The Network File System*, covers the most popular system for sharing files among UNIX and Linux systems. However, UNIX systems also need to share files with systems, such as Windows, that don’t natively support NFS. Enter SMB.

In the early 1980s, Barry Feigenbaum created the BAF protocol to afford shared network access to files and resources. Before release, the name was changed from the author’s initials to Server Message Block (SMB). The protocol was rapidly embraced by Microsoft and the PC community because it gave “just like local” access to files on remote systems.

In 1996, a version called the Common Internet File System (CIFS) was released by Microsoft, mostly as a marketing exercise. Sun Microsystems had also entered the fray in 1996 with its WebNFS offering, and Microsoft saw an opportunity to market SMB with a more user-friendly implementation and name. CIFS introduced (often-buggy) changes to the original SMB protocol. As a result, Microsoft released SMB 2.0 in 2006 and then SMB 3.0 in 2012. Although it’s common within the industry to refer to SMB fileshares as CIFS, the truth is that CIFS was deprecated long ago; only SMB lives on.

If you’re working in a homogeneous UNIX and Linux environment, then this chapter probably isn’t for you. But if you need a way to share files between UNIX and Windows systems, read on.
22.1 Samba: SMB Server for UNIX

Samba is a popular software package, available under the GNU Public License, that implements the server side of the SMB protocol on UNIX and Linux hosts. It was originally created by Andrew Tridgell, who first reverse-engineered the SMB protocol and published the resulting code in 1992. Here, we focus on Samba version 4.

Samba is well supported and under active development to expand its functionality. It offers a stable, industrial strength way to share files between UNIX and Windows systems. The real beauty of Samba is that you install only one package on the server side; no special software is needed on the Windows side.

In the Windows world, a filesystem or directory made available over the network is known as a “share.” It sounds a bit strange to UNIX ears, but we follow this convention when referring to SMB filesystems.

Although we explore only file sharing in this chapter, Samba can also implement a variety of other cross-platform services, including:

- Authentication and authorization
- Network printing
- Name resolution
- Service announcement (file server and printer “browsing”)

Samba can also perform the basic functions of a Windows Active Directory controller. This configuration involves a certain amount of hubris, though; we suspect that being an AD controller is probably a job best left to Windows servers.

There is certainly value in getting your UNIX and Linux systems added to an AD domain as clients, however. This arrangement lets you share identity and authentication information site-wide. See Chapter 17, Single Sign-On, for more information.

Likewise, we don’t recommend Samba as a print server. CUPS is probably your best bet there. See Chapter 12 for more information about printing in UNIX and Linux with CUPS.

Most of Samba’s functionality is implemented by two daemons, smbd and nmbd. smbd implements file and print services as well as authentication and authorization. nmbd is responsible for the other major SMB components: name resolution and service announcement.

Unlike NFS, which requires kernel-level support, Samba requires no drivers or kernel modifications and runs entirely as a user process. It binds to the sockets used for SMB requests and waits for a client to request access to a resource. Once a request has been authenticated, smbd forks an instance of itself that runs as the user who is making the requests. As a result, all normal file access permissions (including group permissions) are obeyed. The only special
functionality that **smbd** adds on top of this is a file locking service that gives Windows systems the locking semantics to which they are accustomed.

If you’re left wondering why you’d use SMB over, say, a more UNIX-integrated remote filesystem such as NFS, the answer is ubiquity. Almost all OSs support SMB at some level. [Table 22.1](#) summarizes some of the main differences between SMB and NFS.

**Table 22.1: SMB vs. NFS**

<table>
<thead>
<tr>
<th>SMB</th>
<th>NFS</th>
</tr>
</thead>
<tbody>
<tr>
<td>User-space servers and processes</td>
<td>Kernel server with threads</td>
</tr>
<tr>
<td>Per-user server processes</td>
<td>Same server (one process) for all clients</td>
</tr>
<tr>
<td>Uses underlying OS for access control</td>
<td>Has its own access control system</td>
</tr>
<tr>
<td>Mounters: usually individual users</td>
<td>Mounters: usually systems</td>
</tr>
<tr>
<td>Pretty good performance</td>
<td>Best performance</td>
</tr>
</tbody>
</table>

[Chapter 21](#) explores NFS in more detail.
22.2 Installing and Configuring Samba

Samba is available for all our example systems. Most Linux distributions include it by default. Patches, documentation, and other goodies are available from samba.org. Make sure you are using the most current Samba packages available for your system, since many updates fix security vulnerabilities.

If Samba is not already installed on your system, you can install it on FreeBSD with `pkg install samba44`. On Linux systems, grab the `samba-common` package through your package manager of choice.

You configure Samba in the `/etc/samba/smb.conf` file (`/usr/local/etc/smb4.conf` on FreeBSD). The file specifies the directories to share, their access rights, and Samba’s general operational parameters. Linux packages are kind enough to supply a heavily commented sample configuration that’s a good starting point for new setups.

Samba comes with sensible defaults for its configuration options, and most sites need only a small configuration file. Run the command `testparm -v` for a listing of all the Samba configuration options and the values to which they are currently set. This listing includes your settings from the `smb.conf` or `smb4.conf` file as well as any default values you have not overridden. Note that once Samba is running, it checks its configuration file every few seconds and loads any changes—no restart required!

The most common use of Samba is to share files with Windows clients. Access to these shares must be authenticated through a user account by one of two options. The first option uses local accounts, for which users specify a password that is managed separately from their other accounts (such as their domain login). The second option integrates Active Directory authentication and so piggybacks on the user’s domain login credentials.
File sharing with local authentication

The simplest way to authenticate users who want to access Samba shares is by creating a local account for them on the UNIX or Linux server.

Because Windows passwords work quite differently from UNIX passwords, Samba cannot control access to SMB shares by means of users’ existing account passwords. Hence, to use local accounts, you must store (and maintain) a separate SMB password hash for every user.

Sometimes, however, simplicity outweighs user convenience, and this authentication system is really simple. Here’s the start of an example `smb.conf` file that uses it:

```
[global]
workgroup = ulsah
security = user
netbios name = freebsd-book
```

The `security = user` parameter tells Samba to use local UNIX accounts. Be sure the workgroup name is set to fit your environment. This is typically the Active Directory domain if you’re in a Windows environment. If you’re not, you can omit this setting.

Samba has its own command, `smbpasswd`, for setting up Windows-style password hashes. For example, here we add the user tobi and set a password for him:

```
$ sudo smbpasswd -a tobi
New SMB password: <password>
Retype new SMB password: <password>
```

The UNIX account should already exist before you attempt to set its Samba password. Users can change their Samba password by running `smbpasswd` without any options:

```
$ smbpasswd
New SMB password: <password>
Retype new SMB password: <password>
```

This example changes the Samba password of the current user on the Samba server. Unfortunately, Windows-only users must log in to a shell prompt on the server to change their share password. The ability to log in remotely must be set up separately, most likely through SSH.
File sharing with accounts authenticated by Active Directory

As simple as the basic process is, maintaining a separate authentication database for shares with `smbpasswd` does seem archaic in today’s hyper-integrated world. In most cases, you’ll want users to authenticate through some form of centralized authority such as Active Directory or LDAP.

Recent years have brought great advances to UNIX and Linux in this area. Chapter 17, Single Sign-On, covers the necessary components, including directory services, sssd, the `nsswitch.conf` file, and PAM. Once you have deployed those components, configuring Samba to take advantage of them is easy. (Historically, `winbind` was used integrate Active Directory with Samba. These days, sssd is the preferred method.)

Here’s an example of the start of an `smb.conf` file for an environment in which Active Directory performs user authentication (via sssd):

```
[global]
    workgroup = ulsah
    realm = ulsah.example.com
    security = ads
    dedicated keytab file = FILE:/etc/samba/samba.keytab
    kerberos method = dedicated keytab
```

In this case, the `realm` parameter should be the same as the local Active Directory domain name. The `dedicated keytab file` and `kerberos method` parameters enable Samba to work properly with Active Directory’s Kerberos implementation.

The keytab file is created by sssd if you’ve set it up according to the instructions in Chapter 17. For more information about keytabs in Samba, see goo.gl/ZxCUKA (deep link within wiki.samba.org).
Configuring shares

After you’ve configured Samba’s general settings and authentication, you can specify in the `smb.conf` file which directories should be shared through SMB. Each share that you expose needs its own stanza in the configuration file. The name of the stanza becomes the share name that is advertised to SMB clients.

Here’s an example:

```
[bookshare]
path = /storage/bookshare
read only = no
```

Here, SMB clients see a mountable share named `\sambaserver\bookshare`. It yields access to the file tree located at `/storage/bookshare` on the server.

Sharing home directories

You can automatically convert users’ home directories into distinct SMB shares with the magic stanza name `[homes]` in the `smb.conf` file:

```
[homes]
comment = Home Directories
browseable = no
valid users = %S
read only = no
```

For example, this configuration would allow user janderson to access her home directory through the path `\sambaserver\janderson` from any Windows system on the network.

At some sites, the default permissions on home directories let users browse one another’s files. Because Samba relies on UNIX file permissions to implement access control, Windows users coming in through Samba can then read one another’s home directories, too. However, experience shows that this behavior tends to confuse Windows users and make them feel exposed.

The variable `%S` listed as the value of `valid users` in the example above expands to the username associated with each share; it thus restricts access to the owner of the home directory. Omit this line if that is not the behavior you want.

Samba uses its magic `[homes]` section as a last resort. If a particular user’s home directory has an explicitly defined share in the configuration file, the parameters set there override the values set through `[homes]`.

Sharing project directories
Samba can map Windows access control lists (ACLs) to either traditional UNIX file permissions or ACLs, if the underlying filesystem supports them. But in practice, we find that ACLs are too complex for most users to deal with.

See this page for more information about ACLs.

Instead of using ACLs, we normally set up a special share for each user group that needs a collective work area. When a user attempts to mount this share, Samba checks that the applicant is in the appropriate UNIX group before allowing access. In the example below, a user must be a part of the eng group to mount the share and access files:

```
[eng]
comment = Group Share for engineering
; Everybody who is in the eng group may access this share.
; People will have to log in with their Samba account.
valid users = @eng
path = /home/eng

; Disable NT ACLs since we do not use them here.
nt acl support = no

; Make sure that all files have sensible permissions and that dirs
; have the setgid (inherit group) bit set.
create mask = 0660
directory mask = 2770
force directory mode = 2000
force group = eng

; Normal share parameters.
browsable = no
read only = no
guest ok = no
```

This configuration does not require you to create a pseudo-user to act as the owner of the shared directory. You just need a UNIX group (here, eng) that includes the intended users of the share.

Users mount the share under their own accounts, but to facilitate collaboration, we would prefer that any files created within the share be owned by group eng. That way, other team members can access newly created files by default.

The first step toward ensuring this behavior is to use the `force group` option to coerce mounters’ effective group IDs to eng, the UNIX group that controls access to the share. However, this step alone is not enough to ensure that new files and directories are assigned a group owner of eng.

As explained here, the setgid option on a directory makes new files created within that directory inherit the directory’s group owner—or at least, it does so on Linux. (FreeBSD does not honor
the setgid bit on a directory; however, its default behavior is to inherit the group, just as Linux does with the setgid bit turned on. Setting the setgid bit on FreeBSD does no harm, however.

We can ensure that new files are owned by eng by setting the group of the share’s root to eng and then turning on the setgid bit on that directory:

```
$ sudo chown root:eng /home/eng
$ sudo chmod u=rwx,g=rwx,o= /home/eng
```

These measures are sufficient to manage files created in the root of the share. However, to make the system work for complex hierarchies of files, we also need to ensure that newly created directories’ setgid bits are also turned on. The example configuration above implements this requirement with the force directory mode and directory mask options.
22.3 Mounting SMB file shares

Mounting for SMB file shares works quite differently from how it’s done for other network filesystems. In particular, SMB volumes are mounted by a specific user rather than being mounted by the system itself.

You need local permission to perform an SMB mount. You also need the password for an identity that the remote SMB server will allow to mount the share. A typical command line on Linux is

```bash
$ sudo mount -t cifs -o username=joe //redmond/joes /home/joe/mnt
```

And the equivalent on FreeBSD:

```bash
$ sudo mount -t smbfs //joe@redmond/joes /home/joe/mnt
```

Windows conceptualizes network mounts as being established by a particular user (hence the `username=joe` option above), whereas UNIX regards them as more typically belonging to the system as a whole. Windows servers generally cannot deal with the concept that several different people might be accessing a mounted Windows share.

From the perspective of the UNIX client, all files in the mounted directory appear to belong to the user who mounted it. If you mount the share as root, then all files belong to root, and garden-variety users might not be able to write files on the Windows server.

The mount options `uid`, `gid`, `fmask`, and `dmask` let you tweak these settings so that ownership and permission bits are more in tune with the intended access policy for that share. Check the `mount.cifs` (Linux) or `mount_smbfs` (FreeBSD) man page for more information about these options.
22.4 BROWSING SMB FILE SHARES

Samba includes a command-line utility called `smbclient` that lets you list file shares without actually mounting them. It also defines an FTP-like interface for interactive access. This feature can be useful when you are debugging or when a script needs access to a share.

For example, here’s how to list shares available to user dan on the server hoarder:

```
$ smbclient -L //hoarder -U dan
Enter dan’s password: <password>
Domain=[WORKGROUP] OS=[Unix] Server=[Samba 3.6.21]

<table>
<thead>
<tr>
<th>Sharename</th>
<th>Type</th>
<th>Comment</th>
</tr>
</thead>
<tbody>
<tr>
<td>Temp</td>
<td>Disk</td>
<td>Temp Storage</td>
</tr>
<tr>
<td>Programs</td>
<td>Disk</td>
<td>Various Programs and Applications</td>
</tr>
<tr>
<td>Docs</td>
<td>Disk</td>
<td>Shared Documents</td>
</tr>
<tr>
<td>Backups</td>
<td>Disk</td>
<td>Backups of all sorts</td>
</tr>
</tbody>
</table>
```

To connect to a share and transfer files, omit the `-L` flag and include the share name:

```
$ smbclient //hoarder/Docs -U dan
```

Once you’re connected, type `help` for a list of available commands.
22.5 Ensuring Samba Security

It’s important to be aware of the security implications of sharing files and other resources over a network. For a typical site, you need to do two things to ensure a basic level of security:

- Explicitly specify which clients can access the resources shared by Samba. This part of the configuration is controlled by the `hosts allow` clause in the `smb.conf` file. Make sure that it contains only the IP addresses, address ranges, or hostnames that it should.

  You can include a `hosts deny` clause in the `smb.conf` file as well, but note that denials have priority. If you include a hostname or address in both the `hosts deny` clause and the `hosts allow` clause, that host will not be able to access the resource.

- Block access to the server from outside your organization. Samba uses encryption only for password authentication. It does not use encryption for its data transport. In almost all cases, you should block access from outside your organization to prevent users from accidentally downloading files in plain text across the Internet.

  Blocking is typically implemented at the network firewall level. Samba uses UDP ports 137–139 and TCP ports 137, 139, and 445.

Since the release of Samba version 3, excellent security documentation has been available on Samba’s wiki, wiki.samba.org.
22.6 Debugging Samba

Samba usually runs without requiring much attention. If you do experience a problem, you can consult two primary sources of debugging information: the `smbstatus` command and Samba’s logging facilities.
Querying Samba’s state with `smbstatus`

`smbstatus` shows currently active connections and locked files; it’s the first place to look when issues arise. This information is especially useful for tracking down locking problems (e.g., “Which user has file xyz open read/write exclusive?”).

```
$ sudo smbstatus
# Some output condensed for clarity
Samba version 4.3.11-Ubuntu

PID  Username  Group  Machine
----------------------------------
6130  clay     atrust  192.168.20.48
23006 dan     atrust  192.168.20.25

Service  pid  machine  Connected at
----------------------------------
admin  6130  192.168.20.48  Wed Apr 12 07:25:15 2017
swdepot2  6130  192.168.20.48  Wed Apr 12 07:25:15 2017
clients  6130  192.168.20.48  Wed Apr 12 07:25:15 2017
clients  23006  192.168.20.25  Fri Apr 28 14:32:25 2017

Locked files:

<table>
<thead>
<tr>
<th>Pid</th>
<th>Uid</th>
<th>DenyMode</th>
<th>R/W</th>
<th>Oplock</th>
<th>SharePath</th>
<th>Name</th>
</tr>
</thead>
<tbody>
<tr>
<td>23006  1009</td>
<td>DENY_NONE</td>
<td>ROONLY</td>
<td>NONE</td>
<td>/atrust/clients</td>
<td>Acme_Supply/Con...</td>
<td></td>
</tr>
<tr>
<td>6130  1035</td>
<td>DENY_ALL</td>
<td>ROONLY</td>
<td>NONE</td>
<td>/home/clay</td>
<td>.</td>
<td></td>
</tr>
<tr>
<td>6130  1035</td>
<td>DENY_NONE</td>
<td>ROONLY</td>
<td>NONE</td>
<td>/atrust/admin</td>
<td>New Hire Proces...</td>
<td></td>
</tr>
</tbody>
</table>
```

The first section of output lists the users that have connected. The Service column in the next section shows the actual shares they’ve mounted. The last section, from which we’ve removed a couple of columns to save space, lists any active file locks.

If you kill the `smbd` associated with a certain user, all that user’s locks disappear. Some applications handle this situation gracefully and reacquire any locks they need. Others freeze and die a horrible death, with much clicking required on the Windows side just to close the unhappy application. As dramatic as this may sound, we have yet to see any file corruption resulting from such a procedure.

Be careful when Windows claims that files have been locked by another application; it is often right. Fix the problem on the client side by closing the offending application instead of brute-forcing the locks on the server.
Configuring Samba logging

Configure logging parameters in your `smb.conf` file:

```
[global]
# The %m causes a separate file to be written for each client.
log file = /var/log/samba.log.%m
max log size = 10000

# If you want Samba to log only through syslog then set the following
# parameter to 'yes'.
syslog only = no

# We want Samba to log a minimal amount of information to syslog.
# Everything should go to /var/log/samba/log.{smbd,nmbd} instead.
# If you want to log through syslog, increase the following parameter
syslog = 7
```

Higher log levels produce more information. Logging uses system resources, so don’t ask for too much detail unless you are actively debugging.

The following example shows log entries generated by an unsuccessful connection attempt:

```
[2017/04/30 08:44:47.510724, 2, pid=87498, effective(0, 0), real(0, 0), class=auth] ..source3/auth/
  auth.c:315(auth_check_ntlm_password)
check_ntlm_password: Authentication for user [dan] -> [dan] FAILED
  with error NT_STATUS_WRONG_PASSWORD
[2017/04/30 08:44:47.510821, 3] ..source3/smbd/
  error.c:82(error_packet_set)
NT error packet at ..source3/smbd/sesssetup.c(937) cmd=115
  (SMBsesssetupX) NT_STATUS_LOGON_FAILURE
```

A successful attempt looks like this:

```
[2017/04/30 08:45:30.425699, 5, pid=87502, effective(0, 0), real(0, 0), class=auth] ..source3/auth/
  auth.c:292(auth_check_ntlm_password)
check_ntlm_password: PAM Account for user [dan] succeeded
[2017/04/30 08:45:30.425864, 2, pid=87502, effective(0, 0), real(0, 0), class=auth] ..source3/auth/
  auth.c:305(auth_check_ntlm_password)
check_ntlm_password: authentication for user [dan] -> [dan]
  -> [dan] succeeded
```

The `smbcontrol` command is handy for altering the debug level of a running Samba server without altering the `smb.conf` file. For example,
This command sets the global debug level to 4 and sets the debug level for authentication-related matters to 10. The `smbd` argument specifies that all `smbd` daemons on the system should have their debug levels set. To debug a specific established connection, use the `smbstatus` command to determine which `smbd` daemon handles the connection, then pass its PID to `smbcontrol` to debug just that one connection. At log levels over 100, you’ll start to see encrypted passwords in the logs.

```
$ sudo smbcontrol smbd debug "4 auth:10"
```
Managing character sets

Starting with version 3.0, Samba encodes all filenames in UTF-8. If your server runs with a UTF-8 locale, which we recommend, this a great match. Type `echo $LANG` to see if your system is running in UTF-8 mode.

If you are in Europe and are still using one of the ISO 8859 locales on the server, you may find that Samba-created filenames that include accented characters (e.g., ä, ö, ü, é, or è) do not display correctly when you run `ls`. The solution is to tell Samba to use the same encoding as your server:

```
unix charset = ISO8859-15
display charset = ISO8859-15
```

Make sure the filename encoding is correct right from the start. Otherwise, files with improperly encoded names accumulate. Fixing them later is a surprisingly complex task.
22.7 RECOMMENDED READING


SAMBA PROJECT. *Samba Wiki Page.* wiki.samba.org. This wiki is updated relatively frequently and is an authoritative source of information, though portions are somewhat disorganized.
A longstanding tenet of system administration is that changes should be structured, automated, and applied consistently among machines. But that’s easier said than done when you’re confronted with a heterogeneous fleet of systems and networks in various states of health.

Configuration management software automates the management of operating systems on a network. Administrators write specifications that describe how servers should be configured, and the configuration management software then brings reality into conformance with the specifications. Several open source implementations of configuration management are in widespread use. In this chapter, we introduce the basics of configuration management and describe the major players.

As an automation tool, configuration management is closely affiliated with the DevOps philosophy of IT operations, which we describe in more detail starting here. People sometimes conflate DevOps and configuration management, so you may occasionally hear these terms used interchangeably. Nevertheless, they are distinct. In this chapter, we show several examples that demonstrate how configuration management enables—but not is not identical to—several key elements of DevOps.

“Configuration management system” is a bit of a handful to read and write, so we often shorten that term to “CM system” (or even simply CM). (Unfortunately, the abbreviation CMS is already in widespread use for “content management system.”)
23.1 Configuration management in a nutshell

The traditional approach to sysadmin automation is an intricate complex of home-grown shell scripts supplemented by ad hoc fire fighting when scripts fail. This scheme works about as well as you might expect. Over time, systems managed in this way usually degenerate into a chaotic wreckage of package versions and configurations that can’t be reliably reproduced. It’s sometimes called the snowflake model of system administration because no two systems are ever alike.

See Chapter 7 for more information about shell scripting.

Configuration management is a better approach. It captures desired state in the form of code. Changes and updates can then be tracked over time in a version control system, which creates an audit trail and a point of reference. The code also acts as informal documentation of a network. Any administrator or developer can read the code to determine how the system is configured.

When all a site’s servers are under configuration management, the CM system effectively acts as both an inventory database and a command-and-control center for the network. CM systems also offer “orchestration” features, which let you apply changes and run ad hoc commands remotely. You can target groups of hosts whose hostnames match particular patterns or whose configuration variables match a given set of values. Managed clients report information about themselves to the central database for analysis and monitoring.

Most configuration management “code” uses a declarative—as opposed to procedural—idiom. Rather than writing scripts that tell the system what changes to make, you describe the state you want to achieve. The configuration management system then uses its own logic to adjust target systems as necessary.

Ultimately, the job of a CM system is to apply a series of configuration specifications, aka operations, to an individual machine. Operations vary in granularity, but they are typically coarse enough to correspond to items that might plausibly appear on a sysadmin’s to-do list: create a user account, install a software package, and so on. A subsystem such as a database might require anywhere between 5 and 20 operations to fully configure. Full configuration of a freshly booted system might entail dozens or hundreds of operations.
23.2 Dangers of Configuration Management

Configuration management is a major improvement over the ad hoc approach, but it is not a magic wand. A few sharp edges are particularly important for administrators to be aware of in advance.

Although all major CM systems use similar conceptual models, they describe these models with different lexicons. Unfortunately, the terminology used by a particular CM system often has more to do with conforming to a marketing theme than with maximizing clarity.

The result is a general lack of conformity and standardization among systems. Most administrators will encounter several CM systems throughout the course of their careers and will develop preferences derived from that experience. Unfortunately, knowledge of one system is not directly portable to another.

As a site grows, so too must the infrastructure needed to support its configuration management system. A site with a few thousand servers needs a handful of systems dedicated to running the CM workloads. This overhead imposes both direct and indirect costs in the form of hardware resources and ongoing maintenance. CM system upgrades can be major projects in their own right.

A certain level of operational maturity and rigour is necessary for a site to fully embrace configuration management. Once a host is under the control of a CM system, it must not be modified manually, or it immediately reverts to the status of a snowflake system.

On more than one occasion, we have seen cases in which hurried (or lazy) administrators have manually updated a configuration-managed host and set their changes to be immutable, thus overriding the expected state and preventing the CM system from applying future changes. This kind of hack results in great confusion when an admin’s colleagues cannot quickly determine why the expected configuration is not being applied. In one case, it caused a major service outage.

Although some CM systems are easier to pick up than others, they are all notorious for having a steep learning curve, especially for administrators who lack prior experience with automation. If you match this description, consider practicing on a lab of virtual machines to hone your skills before you tackle your production network.
23.3 Elements of Configuration Management

In this section, we review the components of a CM system and the concepts used to configure it at a greater level of detail. Then, starting here, we survey four of the most commonly used CM systems: Ansible, Salt, Puppet, and Chef.

Rather than adopt any particular CM system’s terminology, we use the clearest and most directly descriptive term we can find for each concept. Table 23.2 maps the correspondences between our vocabulary and those of the four CM systems listed above. If you’re already familiar with one of those CM systems, you might find it helpful to refer to this table as you read the material below.
**Operations and parameters**

We’ve already introduced the concept of operations, which are the small-scale actions and checks used by a CM system to achieve a particular state. Every CM system includes a large set of supported operations, and more arrive with each new release.

Here are some sample operations that all CM systems can handle right out of the box:

- Create or remove a user account or set its attributes
- Copy files to or from the system being configured
- Synchronize directory contents
- Render a configuration file template
- Add a new line in a configuration file
- Restart a service
- Add a **cron** job or **systemd** timer
- Run an arbitrary shell command
- Create a new cloud server instance
- Create or remove a database account
- Set database operating parameters
- Perform Git operations

This is just a sampling; most CM systems define hundreds of operations, including many that perform potentially complex niche operations, such as setting up specific databases, run-time environments, or even pieces of hardware.

If operations seem suspiciously similar to shell commands, your intuition is accurate. They are scripts, usually written in the implementation language of the CM system itself and exploiting the system’s standard tools and libraries. In many cases, they run standard shell commands under the hood as part of their implementation.

Just as UNIX commands accept arguments, most operations accept parameters. For example, a package management operation would accept parameters that specify the package name, version, and whether the package is to be installed or removed.

Parameters vary according to the operation. As a convenience, they usually have default values that are suitable for the most common use cases.
CM systems let you use variable values (see the next section) to define parameters. They can also infer parameter values according to the environment of the system, such as the network it lives on, whether a particular configuration property is present, or whether the system’s hostname matches a given regular expression.

A well-behaved operation knows nothing about the host or hosts to which it might eventually be applied. The implementation is written to be relatively generic and OS-agnostic. Binding operations to specific systems occurs at a higher level of the configuration management hierarchy.

Despite CM systems’ focus on declarative configuration, operations must ultimately run like any other command. Execution has a start and an end. It can succeed or fail. It reports its status back to the calling environment.

However, operations differ from typical UNIX commands in a few important ways:

- Most operations are designed to be applied repeatedly without causing problems. Borrowing a term from linear algebra, you’ll sometimes see this latter property referred to as “idempotence.”
- Operations know when they change the system’s actual state.
- Operations know when the system state needs to be changed. If the current configuration already conforms to the specification, the operation exits without doing anything.
- Operations report their results to the CM system. Their report data is richer than a UNIX-style exit code and can aid in debugging.
- Operations strive to be cross-platform. They usually define a constrained set of functions that are common to all supported platforms, and they interpret requests in accordance with the local system.

Some operations can’t be made idempotent without a little help from a sysadmin who knows more about the context. For example, if an operation runs a garden-variety UNIX command, the CM system has no direct way of knowing what effect that command had on the system.

You also have the option of writing your own custom operations. They’re just scripts, and the CM system typically provides a well-greased path for integrating your custom operations with the standard ones.
Variables

Variables are named values that influence how configurations are applied to individual machines. They commonly set parameter values and fill in the blanks in configuration templates.

Variable management in CM systems is often quite rich. A few points of note:

- Variables can typically be defined in many different places and contexts within the configuration base.
- Each definition has a scope in which it’s visible. Scope types vary by CM system and might encompass a single machine, a group of machines, or a particular set of operations.
- Multiple scopes can be active in any given context. Scopes can be nested, but more commonly they are simply coactive.
- Because multiple scopes can define values for the same variable, some form of conflict resolution is necessary. Some systems merge values, but most use precedence rules or definition order to pick a winning value.

Variables are not limited to having scalar values; arrays and hashes are also acceptable variable values in all CM systems. Some operations accept nonscalar parameter values directly, but such values are more typically used above the level of individual operations. For example, an array might be enumerated in a loop to apply the same operation more than once with different parameters.
**Facts**

CM systems investigate each configuration client to determine descriptive facts such as the IP address of the primary network interface and the OS type. This information is then accessible from within the configuration base through variable values. As with any other variable, these values can be used to define parameter values or to expand templates.

It can take awhile to determine all the facts associated with a particular system. Therefore, CM systems generally cache facts, and they do not necessarily rebuild the cache on every run. If you find that a particular configuration flow is encountering stale configuration data, you might need to explicitly invalidate the cache.

All CM systems let target machines add their own values to the fact database, either by including a static file of declarations or by running custom code on the target machine. This feature is useful both for extending the types of information that can be accessed through the facts database and for moving static configuration information onto client machines.

Client-side hints can be particularly useful for managing cloud and virtual servers. You simply apply cloud-level markers (such as EC2 tags) as an instance is created, and the configuration management system can then flesh out the appropriate configuration by checking the markers. Keep in mind the security implications of this approach, however: the client controls the facts that it reports, so make sure that a compromised client can’t exploit the configuration management system to gain additional privileges.

Depending on the CM system, you may be able to transcend your local context when sniffing around in the variable or fact space. In addition to accessing the configuration information for the current host, you may also be able to access data for other hosts, or even to introspect the state of the configuration base itself. This is a useful feature for coordinating a distributed system such as a cluster of servers.
Change handlers

If you change a web server’s configuration file, you had better restart the web server. That’s the basic concept behind handlers, which are operations that run in response to some sort of event or situation rather than as part of a baseline configuration.

In most systems, a handler runs whenever one or more of a designated set of operations reports that it has modified the target system. The handler isn’t told anything about the exact nature of the change, but because the association between operations and their handlers is fairly specific, additional information isn’t needed.
**Bindings**

Bindings complete the basic configuration model by associating specific sets of operations to specific hosts or groups of hosts. You can also bind operations to a dynamic set of clients that’s defined by the value of a fact or variable. CM systems can also define host groups by looking up information in a local inventory system or by calling a remote API.

In addition to their basic linking role, bindings in most CM systems also act as variable scopes. This feature lets you customize the behavior of the operations you’re assigning by defining or customizing variable values for the clients that are targeted.

A given host can match criteria for many different bindings. For example, a node might live on a certain subnet, be managed by a particular department, or fill an explicitly designated role (e.g., Apache web server). The CM system takes account of all of these factors and activates the operations associated with each binding.

Once you set up the bindings for a host, you can invoke your CM system’s top-level “configure everything” mechanism to make the CM system identify all the operations that should run on the target and execute them in order.
Bundles and bundle repositories

A bundle is a collection of operations that perform a specific function, such as installing, configuring, and running a web server. CM systems let you package bundles into a format that’s suitable for distribution or reuse. In most cases, a bundle is defined by a directory, and the name of the directory defines the name of the bundle.

CM vendors maintain public repositories that include both officially blessed and user-contributed bundles. You can use these “as is” or modify them to suit your needs. Most CM systems provide native commands for interacting with repositories.
Environments

It’s often useful to segregate configuration-managed clients into multiple “worlds,” such as the traditional categories of development, test, and production. Large installations can create even finer distinctions to support processes such as the gradual (“staged”) rollout of new code into production.

These different worlds are known generically as “environments,” both inside and outside the configuration management context. This seems to be the single term on which all configuration management systems agree.

When properly implemented, environments are not just groups of clients. They’re an additional axis of variation that can affect multiple aspects of the configuration. The development and production environments might both include web servers and database servers, for example, but the details of how those roles are defined might vary among environments.

For example, it’s common for the database and web server to run on the same machine in the development environment. However, the production environment usually has multiple servers of each type. A production environment might also define server types that don’t exist in the development environment, such as those that do load balancing or act as DMZ proxies.

The environment system is usually thought of as a sort of pipeline for configuration code. As a thought experiment, you can imagine that fixed groups of clients run the development, test, and production environments. As a given configuration base is validated, it propagates from one environment to the next, ensuring that changes are properly vetted before they reach the all-important production systems.

On most CM systems, different environments are just different versions of the same configuration base. If you’re a Git user, think of them as tags in a Git repository: the development tag points to the most recent version of the configuration base, and the production tag might point to a commit that’s several weeks old. The tags move forward as releases make their way through testing and deployment.

Different environments can provide different variable values to clients. For example, the database credentials used in development likely differ from those used on production systems, as do the details of the network configuration and perhaps the users and groups who are permitted access.

See Chapter 26, Continuous Integration and Delivery, for more information about environments.
Client inventory and registration

Because CM systems define lots of ways to segregate clients into categories, the overall universe of machines under configuration management must be well defined. The inventory of managed hosts can live in a flat file or in a proper relational database. In some cases, it may even be entirely dynamic.

The exact mechanism through which configuration code is distributed, parsed, and executed varies among CM systems. Most systems actually give you several options in this regard. Here are a few common approaches:

- A daemon runs continuously on each client. The daemon pulls its configuration code from a designated CM server (or server group).
- A central CM server pushes configuration data to each client. This process can run on a regular schedule, or it can be initiated by administrators.
- Each managed node runs a client that wakes up periodically, reads configuration data from a local clone of the configuration base, and applies the relevant configuration to itself. There is no central configuration server.

Configuration information is sensitive and often includes secrets such as passwords. To protect this data, all CM systems define some way for clients and servers to authenticate each other and to encrypt private information.

The process of putting a new client under configuration management can be made as simple as installing the appropriate client-side software. If the environment has been configured to support automatic bootstrapping, a new client can automatically contact its configuration server, authenticate itself, and initiate the configuration process. OS-specific initialization mechanisms typically kick off this chain of events the first time the client is bootstrapped. The flow is depicted in Exhibit A.

Exhibit A: Initialization process for a new CM-managed client
New client first boot
rule = web
environment = production

Client downloads and installs agent

CM agent automatically authenticates and registers with the CM server

Client enters operation, leaving agent running as a background daemon

CM agent applies the list of bindings

Server identifies the bindings for a production web server
23.4 **Popular CM systems compared**

Currently, four major players own the market for general configuration management on UNIX and Linux systems: Ansible, Salt, Puppet, and Chef. **Table 23.1** shows some general information about these packages.

**Table 23.1: Major configuration management systems**

<table>
<thead>
<tr>
<th>System</th>
<th>Web site</th>
<th>Impl</th>
<th>Config</th>
<th>Template</th>
<th>Server</th>
<th>Client</th>
</tr>
</thead>
<tbody>
<tr>
<td>Ansible</td>
<td>ansible.com</td>
<td>Python</td>
<td>YAML</td>
<td>Jinja</td>
<td>No</td>
<td>No</td>
</tr>
<tr>
<td>Salt</td>
<td>saltstack.com</td>
<td>Python</td>
<td>YAML</td>
<td>Jinja</td>
<td>Optional</td>
<td>Optional</td>
</tr>
<tr>
<td>Puppet</td>
<td>puppet.com</td>
<td>Ruby</td>
<td>custom</td>
<td>ERB*</td>
<td>Optional</td>
<td>Optional</td>
</tr>
<tr>
<td>Chef</td>
<td>chef.io</td>
<td>Ruby</td>
<td>Ruby</td>
<td>ERB</td>
<td>Optional</td>
<td>Yes</td>
</tr>
</tbody>
</table>

* ERB (embedded Ruby) is a basic syntax for embedding Ruby code in templates.

All of these packages are relatively young. The oldest, Puppet, debuted in 2005. It still claims the largest market share, in large part because of its early head start. Chef was released in 2009, followed by Salt in 2011 and Ansible in 2012.

The general category of configuration management software was pioneered by Mark Burgess’s CFEngine in 1993. CFEngine is still around and is still actively developed, but the majority of its user base has been siphoned away by newer systems. See cfengine.com for current information.

Microsoft has its own CM solution in the form of PowerShell Desired State Configuration. Although it originates in the Windows world and is primarily designed to configure Windows clients, Microsoft has also published extensions for configuring Linux systems. It’s worth noting that all four of the systems in **Table 23.1** can configure Windows clients as well.

A number of projects focus on specific subdomains of configuration management, notably new-system provisioning (e.g., Cobbler) and software deployment (e.g., Fabric and Capistrano). The general proposition behind these systems is that by more closely modeling a specific problem domain, they can provide a simpler and more targeted set of features.

Depending on your needs, you may or may not find that these specialized systems provide a reasonable rate of return on your learning investment. Generic configuration management systems like those in **Table 23.1** are not perfectly suited to all possible activities.

The systems in **Table 23.1** work with pretty much any type of contemporary UNIX-compatible client machine, although there’s always a support frontier. Chef has a modest edge in compatibility and supports even AIX.
See Chapter 25 for more information about containers.

OS support on the configuration server side (for those systems that use a configuration server) is more limited. Chef, for example, requires RHEL or Ubuntu for its server. Containerized versions of the server can run anywhere, though, so this isn’t as much an obstacle as it might seem.
Terminology

Table 23.2 shows the terms used by each of our example CM systems for the entities outlined in *Elements of configuration management*.

<table>
<thead>
<tr>
<th>Our term</th>
<th>Ansible</th>
<th>Salt</th>
<th>Puppet</th>
<th>Chef</th>
</tr>
</thead>
<tbody>
<tr>
<td>operation op type</td>
<td>task</td>
<td>state</td>
<td>resource</td>
<td>resource</td>
</tr>
<tr>
<td></td>
<td>module</td>
<td>function</td>
<td>resource type, provider</td>
<td>provider</td>
</tr>
<tr>
<td>op list parameter binding</td>
<td>tasks</td>
<td>states</td>
<td>class, manifest</td>
<td>recipe</td>
</tr>
<tr>
<td></td>
<td>parameter</td>
<td>parameter</td>
<td>property, attribute classification, declaration</td>
<td>attribute run list</td>
</tr>
<tr>
<td>master host client host client group</td>
<td>control</td>
<td>master</td>
<td>master</td>
<td>server</td>
</tr>
<tr>
<td></td>
<td>host</td>
<td>minion</td>
<td>agent, node group</td>
<td>node role</td>
</tr>
<tr>
<td></td>
<td>group</td>
<td>nodegroup</td>
<td>node group</td>
<td></td>
</tr>
<tr>
<td>variable fact</td>
<td>variable</td>
<td>variable</td>
<td>parameter, variable fact</td>
<td>attribute automatic attribute</td>
</tr>
<tr>
<td></td>
<td>fact</td>
<td>grain</td>
<td>fact</td>
<td></td>
</tr>
<tr>
<td>notification handler</td>
<td>notification</td>
<td>requisite</td>
<td>notify</td>
<td>notifies subscribes</td>
</tr>
<tr>
<td></td>
<td>handler</td>
<td>state</td>
<td>subscribe</td>
<td></td>
</tr>
<tr>
<td>bundle bundle repo</td>
<td>role</td>
<td>formula</td>
<td>module</td>
<td>cookbook</td>
</tr>
<tr>
<td></td>
<td>galaxy</td>
<td>GitHub</td>
<td>forge</td>
<td>supermarket</td>
</tr>
</tbody>
</table>
**Business models**

All the products we discuss are freemium-model packages, which means that the basic systems are open source and free, but that each system has a corporate backer that sells support, consulting services, and add-on packages.

In theory, vendors have a potential motivation to withhold useful functionality from the open source releases to motivate add-on sales. In the configuration management space, however, this effect has not been evident. The open source versions of the software are full-featured and more than adequate for most sites.

Add-on services are mostly of interest to large organizations. If your site falls into this category, you may want to evaluate configuration management systems with respect to the functionality and pricing of the full-stack offerings. The main upsells are usually support, custom development, training, GUIs, and reporting and monitoring solutions. In this book, we discuss only the basic, free versions.
Architectural options

In theory, CM systems don’t require servers. Software could run only on the machines being configured. You’d copy the configuration base to each target host and simply run a command to say, “Here, configure yourself according to these specifications.”

In practice, it’s nice not to have to fuss with the details of getting configuration information pushed out to clients and executed. CM systems always make some provision for centralized control, even if the master machine is defined as “wherever you happen to be logged in and have a clone of the configuration base.”

Ansible uses no daemons at all (other than sshd), which is an appealing simplification. Configuration runs happen when an administrator (or cron job) on the server runs the ansible-playbook command. ansible-playbook executes the appropriate remote commands over SSH, leaving no trace of its presence on the client machine after configuration has completed. The only requirements for client machines are that they be accessible through SSH and have Python 2 installed. (Depending on the system, you might need a Python add-on or two as well. For example, Fedora requires the python-dnf package.)

Salt, Puppet, and Chef include both master- and client-side daemons. Typical deployment scenarios run daemons on both sides of the relationship, and this is the environment you’ll see described in most documentation. It’s possible to run each of these without a server also, but this configuration is less common.

It’s tempting to assume that CM systems with daemons must be more heavyweight and more complex than those without (i.e., Ansible). However, that isn’t necessarily true. In Salt and Puppet, the daemons are facilitators and accelerators. They’re useful but optional, and they don’t change the fundamental architecture of the system, although they do enable some advanced features. If you prefer, you are free to run these systems without daemons and to replicate the configuration base by hand. Salt even has an SSH-based mode that works similarly to Ansible.

Given that, why would you want to mess around with a bunch of optional daemons? Several reasons:

- It’s faster. Ansible works hard to overcome the performance limits imposed by SSH and by its lack of client-side caching, but it is still noticeably more sluggish than Salt. When you’re reading a system administration book, ten seconds sounds insignificant. In the midst of resolving an outage, it feels endless, especially when repeated across dozens or hundreds of clients.
- Some features can’t exist without central coordination. For example, Salt lets clients notify the configuration master of events such as full disks. You can then respond to these events through the normal configuration management facilities. Having a central connection point facilitates a variety of interclient data-sharing features.
• Only the master-side daemon is really a potential source of administrative complexity. CM systems work hard to make client bootstrapping a one-line operation, regardless of whether a daemon is involved.

• The presence of active agents on both client and server opens a variety of architectural options not available in one-sided configurations.

In terms of architecture, Chef is the outlier among configuration management systems in that its server daemon is a top-tier entity within the conceptual model. Salt and Puppet serve configuration data directly from plaintext files on disk; to make changes, you simply edit the files. By contrast, the Chef server is an opaque and authoritative source of configuration information. Changes must be uploaded to the server with the `knife` command or they will not be available to clients. (However, even Chef has a serverless mode of operation in the form of `chef-solo`.)

We mention all this not to promote serverful systems per se, but simply to point out that the main fault line among CM systems runs between Chef and everything else. Ansible, Salt, and Puppet all have about the same, modest level of overall complexity. Chef requires significantly more investment to maintain and master, especially when its extensive line of add-on modules is added to the mix.

Because of its serverless model, Ansible is often tagged as a sort of “easy option” for configuration management. But in fact, the basic architectures of Salt and Puppet are similarly approachable. A strong case could be made that Salt is the simplest system of all, if you disregard its advanced facilities and somewhat peculiar documentation. Don’t write off Salt and Puppet as advanced options.

The converse is also true: Ansible is more than just a gimped-out starter system for short-attention-span sysadmins. It’s a legitimate option for complex sites, although its sluggish performance becomes increasingly more apparent in this context.
Language options

Ansible and Salt are written in Python. Puppet and Chef are written in Ruby. But except in the case of Chef, this information is probably less relevant than it might initially appear.

No Python code appears in the average Ansible or Salt configuration. Both of these systems use YAML (an alternate syntax for expressing JavaScript object notation, aka JSON), as their primary configuration language. YAML is just structured data, not code, so it has no inherent behavior other than the interpretation assigned by the configuration management system.

Here’s a simple example from Salt that keeps the SSH service enabled and running:

```yaml
ssh.server.run_ssh:
  service:
    - name: sshd
    - running
    - enable: true
```

To make YAML files more dynamically expressive, both Ansible and Salt augment them with a templating system, Jinja2, as a preprocessor. Jinja has its roots in Python, but it’s not just a simple Python wrapper. In use, it feels more like a template system than a real programming language. Even Salt, which relies more heavily on Jinja than does Ansible, cautions against putting too much logic into Jinja code.

In fairness, Salt is actually format- and preprocessor-agnostic, and it supports several input pipelines (including raw Python) right out of the box. However, departing from the greased path of Jinja and YAML means leaving the documentation and the rest of the world behind. It’s probably best deferred until you’re quite familiar with Salt.

The bottom line is that unless you write your own custom operation types or use explicit escapes into Python, you won’t be encountering much Python in the Ansible and Salt worlds. Extending the CM system with your own code can in fact be quite easy and quite helpful, however.

Jon Corbet, one of our technical reviewers, agrees that these systems don’t expose much Python…until things go horribly wrong. “At that point,” he adds, “familiarity with Python tracebacks and data structure representations helps a lot.”

Both Puppet and Chef use Ruby-based, domain-specific languages as their primary configuration systems. Chef’s version is a lot like a configuration management analog of Rails from the web development world. That is, it has been extended with a few concepts that are designed to facilitate configuration management, but it’s still recognizably Ruby. For example:

```ruby
service 'sshd' do
  supports :restart => true, :status => true
  action [:enable, :start]
end
```
Most configuration management tasks can be achieved without delving below the surface of Ruby, but Ruby’s full power is available if you need it. You’ll appreciate this hidden depth more and more as your comfort with Ruby and Chef increases.

By contrast, Puppet has put in quite a bit of work to be conceptually independent of Ruby and to use it only as an implementation layer. Although the language remains Ruby under the hood and is amenable to the insertion of Ruby code, the Puppet language has its own idiosyncratic structure that is more akin to a declarative system such as YAML than a programming language:

```ruby
service {
  "ssh":
    ensure => "running",
    enable => "true"
}
```

In our opinion, Puppet hasn’t done administrators any favors with this architecture. Instead of letting you leverage your existing knowledge of Ruby (or parlay your Puppet experience into a more general familiarity with Ruby), Puppet just defines its own insulated world.
Dependency management options

No matter how your configuration management system structures its data, the work list for a given client ultimately boils down to a set of operations for the client to execute. Some of those operations have execution-order dependencies, and some don’t.

For example, consider the following Ansible tasks for installing a www user account, such as might be used to own the files for a web application:

- name: Ensure that www group exists
  
  group: name=www state=present

- name: Ensure that www user exists
  
  user: name=www group=www state=present createhome=no

We want the www user to have its own dedicated group, also named www. Ansible’s user module does not create groups automatically, so we must do that in a separate step. And the group creation must precede the creation of the www account; it’s an error to specify a nonexistent group in a user operation.

Ansible executes operations in the order in which they are presented by the configuration, so this configuration snippet works just fine. Chef works this way, too, in part because it’s much harder to rearrange code than data. Even if it wanted to, Chef couldn’t reliably break your code into pieces and reassemble the pieces as it sees fit.

By contrast, Puppet and Salt allow dependencies to be explicitly declared. For example, in Salt, the equivalent set of states would be

```yaml
www-user:
  user.present:
    - name: www
    - gid: www
    - createhome: false
    - require:
      - www-group

www-group:
  group.present:
    - name: www
```

We inverted the order of the operations here for dramatic effect. But because of the require declaration, the operations run in the correct order regardless of how they appear in the source file. The following command applies the configuration:
The require parameter can be added to any operation (“state,” in Salt) to ensure that the named prerequisites run before the current operation. Salt defines several types of dependency relationships, and declarations can appear on either side of a relationship.

Puppet works similarly. It also helps to ease the pain of declaring dependencies by inferring them automatically in some common circumstances. For example, a user configuration that names a particular group automatically becomes dependent on the resource that configures that group. Nice!

So... Why would you want to declare your dependencies explicitly when configuration order seems natural and effortless? Apparently, lots of administrators have been asking this question, as both Salt and Puppet have moved to a hybrid dependency model in which presentation order is significant. However, it’s only a factor within a given configuration file; inter-file dependencies must still be explicitly declared.

The main benefit of declaring dependencies is that it makes configurations more resilient and explicit. The CM system is not obliged to abort the configuration process at the first sign of trouble, because it knows which subsequent operations might be affected by a failure. It can abort one dependency chain while allowing others to continue. Nice, but in our view not a significant payback for the extra work of declaring dependencies.
In theory, a CM system that knows dependency information can parallelize the execution of independent operation chains on a particular host. However, neither Salt nor Puppet attempts this feat.
General comments on Chef

We’ve seen deployments of the mainline CM packages at organizations of various sizes, and they all display something of a tendency toward entropy. The Ansible access options section includes some hints for keeping things organized. However, an even more fundamental rule is to avoid taking on more complexity than is helpful for your environment.

In practice, this means you need to be clear about whether you’re living in Chef territory or not. Chef thinks big. To get the most out of Chef, you should have

- Hundreds or thousands of machines under configuration management
- An administrative staff of nonuniform privileges and experience (Chef’s internal permissions system and multiple interfaces are quite helpful here)
- Specific reporting, compliance, or regulatory requirements to enforce
- The patience to train new team members without prior Chef experience

Sure, you can run Chef stand-alone on a single machine for free. Nobody’s stopping you! But you’ll still have to pay the cognitive overhead for many of the enterprise-level features you aren’t using. They’re baked into the architecture and the documentation.

We like Chef. It’s complete, robust, and scalable—more so than the alternatives. But at heart, it’s just another configuration management system that does the same basic stuff as Ansible, Salt, and Puppet. Keep Chef in perspective, and resist the temptation to adopt it just “because it’s the most powerful” (or “because it uses Ruby,” for that matter).

We have found that bringing beginners up to speed with Chef can be a significant challenge, especially for those without prior configuration management experience. Chef requires a developer mindset more than the other systems do. Prior programming experience is helpful.

Chef’s attribute precedence system is powerful but can also be a source of frustration. Its peculiar combination of foodie and Internet-meme nomenclature is annoying and uninformative. Resolving dependencies among cookbooks can be challenging; sometimes an upstream dependency breaks, and all your systems develop problems unless you remembered to pin all your dependencies to a particular version.
General comments on Puppet

Puppet is the oldest of the four main CM systems and the one with the largest installed base. It has lots of users, lots of contributed modules, and a free web GUI. Still, it’s losing market share fairly steadily to more recent competitors.

As a determinedly middle-of-the-road option, Puppet is under pressure from both ends of the market. It’s famous for server-side bottlenecks that cause problems when managing thousands of hosts, and several major Puppet deployments have abandoned it over the last couple of years (most publicly, Lyft, which adopted Salt). These days, such large-scale scenarios seem to be better handled with a tiered Chef or Salt network.

In the arena of small deployments, Ansible and Salt are mounting a serious challenge with their relatively low barriers to entry. As discussed on this page, Puppet is not complex at heart. However, it does drag along some historical baggage that tends to impede newcomers. For example, relatively few operations are built into the Puppet core. Most sites will need to go prospecting for user-contributed modules to complete their basic configurations.

Our subjective impression is that Puppet went through some false starts early in its design and development. Although Puppet has worked hard to correct these issues, history and backward compatibility take an inevitable toll on the current product.

It doesn’t help that Puppet transmutes the golden treasure of Ruby into the lump of coal that is the Puppet configuration language. That was probably a sensible decision back in 2005, when Ruby was obscure and Rails had not yet appeared on the scene to propel it to stardom. These days, the Puppet configuration language just seems gratuitous.

None of these issues is a deal breaker, but Puppet seems to have no clear and compelling advantage that might counterbalance such concerns. We are not aware of any bake-off or comparative review written within the last few years in which Puppet emerged as a recommended option.

Of course, if you’ve inherited an existing Puppet installation, there’s no need to start looking for an immediate replacement. Puppet works fine; the distinctions among these systems are mostly a matter of style and marginal advantage.
General comments on Ansible and Salt

Ansible and Salt are both nice systems, and we recommend one of these options for the majority of sites.

We’ve taken a deeper look at both of these systems in Introduction to Ansible and Introduction to Salt, which begin here and here, respectively. Each of those sections reviews the system’s configuration syntax and the general flavor of day-to-day use.

Ansible and Salt look deceptively similar on the surface, mostly because they both use YAML and Jinja as their default formats. Under the hood, however, they almost couldn’t be more different. Accordingly, we defer our head-to-head comparison of Ansible and Salt until this page, once we’ve discussed them both in a bit more detail.

Before we look into the systems themselves, however, we cast a jaundiced eye on YAML itself.
**YAML: a rant**

In theory, a YAML document should start with three dashes on a line by themselves, and the Ansible documentation often follows this convention. However, this “start YAML document” line is essentially vestigial. As far as we are aware, it can safely be omitted in all cases.

As mentioned earlier, YAML is just an alternate syntax for JSON. For example, this YAML for Ansible:

```yaml
- name: Install cpdf on cloud servers
  hosts: cloud
  become: yes
  tasks:
    - name: Install OCAML packages
      package: name= {{ item }} state=present
      with_items:
      - gmake
      - ocaml
      - ocaml-opam
```

maps to the following JSON:

```json
[
  {
    "name": "Install cpdf on cloud servers",
    "hosts": "cloud",
    "become": "yes",
    "tasks": [
      {
        "name": "Install OCAML packages",
        "package": {
          "name": "{{ item }}",
          "state": "present",
        },
        "with_items": [ "gmake", "ocaml", "ocaml-opam" ]
      }
    ]
  }
]
```

In the JSON world, brackets enclose lists and curly braces enclose hashes. A colon separates a hash key from its value. These delimiters can appear directly in YAML, but YAML also understands indentation to indicate structure, much like Python. YAML marks items in a list with a preceding dash.

Take a moment to verify that you understand how the YAML example above maps into JSON, because Ansible and Salt are actually JSON-based worlds. The YAML is just a shorthand. We pick on Ansible below, as its version of YAML is a bit more idiosyncratic, but most of the general points apply to Salt as well. (Once again, Salt partisans will protest that Salt can’t be blamed for YAML and Jinja because it has no actual dependencies on these systems. You’re free
to use any one of a number of alternatives. That’s all true. At the same time, it’s a lot like saying that you’re not responsible for the country’s government because you didn’t vote.)

Clearly, the YAML version is more readable than the JSON version. The problem isn’t YAML per se, but rather the compromises inherent in trying to force data of the complexity found in configuration management systems into the mold of JSON.

YAML is good for representing simple data structures, but it’s not a tool that scales well to arbitrary complexity. When cracks appear in the model, they have to be puttied over with a variety of ad hoc fixes.

The example above already contains such a patch. Did you spot it?

```yaml
package: name={{ item }} state=present
```

Ignore the `{{ item }}` part; that’s just a Jinja expansion. The crime here is the `name=value` syntax, which is really just a nonstandard shorthand for defining a sub-hash:

```yaml
package:
  name: {{ item }}
  state: present
```

Or is it? Actually not, because Ansible doesn’t allow hash values that start with a Jinja expansion. That Jinja term now must sport quotes:

```yaml
package:
  name: "{{ item }}"
  state: present
```

And what if the operation accepts a “free form” argument?

```yaml
- name: Cry for help
  shell: echo "Please, sir, I just want the syntax to be consistent"
  args:
    warn: no
```

Visually, this doesn’t look so bad. But think about what’s actually going on: `shell` is the operation type, and `warn` is a parameter for `shell` just as `state` is a parameter for `package` in the previous example. So what is that extra `args` dictionary doing there?

Well, `shell` typically has a complex string as its main argument (the shell command to run), so it’s been made a special type of operation that accepts a string instead of a parameter hash as its value. The `args` dictionary is actually a property of the task-item wrapper, not the `shell` operation. Its contents are covertly stuffed down into the `shell` operation on your behalf to make the whole construction work.
No problem; keep calm and carry on. But it’s a confusing subtlety that muddles a relatively basic example.

The problem isn’t this specific scenario. It’s the constant drip of edge cases, ambiguities, and compromises that are needed to coerce configuration data into JSON format. Does this particular argument go in the operation? In the state? In the binding? It’s all just a big JSON hierarchy, so the answer is rarely obvious.

Look again at the “Install cpdf on cloud servers” playbook on this page. Is it obvious that with_items should be at the same level as package and not at the same level as name and state (which are in fact logically beneath package)? Probably not.

The underlying intent behind the YAML approach is praiseworthy: use an existing format that people already know, and represent configuration information as data instead of code. Still, these systems have syntactic warts that would probably not be permitted in a real programming language.

Despite the looseness of YAML as used in configuration management systems, its specification is actually quite lengthy. In fact, it’s longer than the specification for the entire Go programming language.
23.5 **Introduction to Ansible**

Ansible has no server daemon and installs no software of its own on clients, so it’s really just a set of commands (most notably, *ansible-playbook*, *ansible-vault*, and *ansible*) that you install on any system from which you wish to manage clients.

*See this page for more information about EPEL.*

Standard OS-level packages are widely available for Ansible, although the package names vary from system to system. On RHEL and CentOS, make sure you have the EPEL repository enabled on the master systems. As with most things, OS-specific packages are often somewhat out of date with respect to the trunk. If you don’t mind forgoing package management, Ansible is easy to install from the GitHub repository (ansible/ansible) or through pip. (pip is a package manager for Python. Try *pip install ansible* to pull the latest version from PyPI, the Python Package Index. You might need to install *pip* from your distribution’s packaging system first.)

The default location of Ansible’s master configuration file is `/etc/ansible/ansible.cfg`. (As with most add-ons, FreeBSD moves the *ansible* directory to `/usr/local/etc`.) The default *ansible.cfg* file is short and sweet. The only change we’d recommend is to add the following lines to the end:

```
[ssh_connection]
pipelining = true
```

*See this page to allow *sudo* without a control terminal.*

These lines turn on pipelining, an SSH feature that significantly improves performance. Pipelining requires that *sudo* on clients not be configured to require interactive terminals; however, that is the default.

If you keep your configuration data under `/etc/ansible`, you’ll need to use *sudo* to make changes, and you tie yourself to one particular server machine. Alternatively, you can easily set up Ansible for use under your own account. The server just runs *ssh* to reach other systems, so root privileges are unnecessary unless you need to run privileged commands on the server side.

Fortunately, Ansible makes it a snap to combine system-wide and personal configurations. Don’t remove the system-wide configuration; just shadow it by creating `~/.ansible.cfg` and setting the location of your inventory file and roles directory:

```
[defaults]
inventory = ./hosts
roles_path = ./roles
```
The inventory is the list of client systems, and roles are bundles that abstract various aspects of client configuration. We return to both of these topics shortly.

Here, we define both locations as relative paths, which assumes that you’ll `cd` to your clone of the configuration base and that you’ll follow the stated naming conventions. Ansible also understands the shell’s `~` notation for home directories if you prefer to use fixed paths. (Ansible allows `~` pretty much everywhere else, too.)
Ansible example

Before we delve into too much more detail, we first look at a small example that demonstrates a few basic Ansible operations.

The following set of steps would set up `sudo` on a new system (as might be required, e.g., on FreeBSD, which does not include `sudo` by default).

1. Install the `sudo` package.
2. Copy a standard `sudoers` file from a server and install it locally.
3. Make sure the `sudoers` file has appropriate permissions and ownerships.
4. Create a UNIX group called “sudo”.
5. Add every system administrator with an account on the local machine to the sudo group.

The Ansible code below implements these steps. Because this code is designed to illustrate several points about Ansible, it isn’t necessarily an example of idiomatic Ansible code.

```
- name: Install sudo package
  package:
    name: sudo
    state: present

- name: Install sudoers file
  template:
    dest: "{{ sudoers_path }}"
    src: sudoers.j2
    owner: root
    group: wheel
    mode: 0640

- name: Create sudo group
  group:
    name: sudo
    state: present

- name: Get current list of usernames
  shell: "cut -d: -f1 /etc/passwd"
  register: userlist

- name: Add administrators to the sudo group
  user:
    name: "{{ item }}"
    groups: sudo
    append: true
  with_items: "{{ admins }}"
  when: "{{ item in userlist.stdout_lines }}"
```

The statements are applied in order, much as they would be in a shell script.

The expressions enclosed in double curly braces (e.g., "{{ admins }}") are variable expansions. Ansible interpolates facts in a similar manner. This kind of parameter management flexibility is a common characteristic of configuration management systems, and it’s one of their main
advantages over raw scripts. You define the general procedure in one place and the configuration specifics elsewhere. The CM system then collapses the global specification and ensures that the proper parameters are applied to each target host.

The file `sudoers.j2` is a Jinja2 template that expands to become the `sudoers` file on the target machine. The template can consist of static text or it can have internal logic and variable expansions of its own.

Templates are usually kept along with configurations in the same Git repository, allowing for one-stop shopping when configurations are applied. There’s no need to maintain a separate file server from which templates can be copied. The configuration management system uses its existing access to the target host to install templates, so credential management need be set up only once.

We had to work around a couple of rough edges. Ansible’s `user` module, used here to add system administrators to the sudo UNIX group, normally ensures that the specified account exists, and it creates the account if it does not. In this scenario, we want to affect only accounts that already exist, so we’re forced to manually check for the existence of each account before we permit `user` to modify it. In a more typical scenario, the configuration management system would be responsible for setting up administrators’ accounts as well as `sudo` access. The configuration specifications for both functions would likely refer to the same `admins` variable, and so there would be no possibility of conflict and no need to validate each account name.

To check for the existence of accounts, the configuration runs the shell command `cut –d: -f1 /etc/passwd` to obtain a list of existing accounts and captures (“registers”) the output under the name `userlist`. It’s similar in principle to the `sh` line

```
userlist=$(cut -d: -f1 /etc/passwd)
```

Each account listed in the `admins` variable (`with_items: "{{ admins }}"`) is considered separately. During its turn, the account name is assigned to the variable `item`. (The name `item` is an Ansible convention; the configuration does not specify it.) For each account found within the output of the `cut` command (the `when` clause), the `user` clause is invoked.

There’s a bit of extra glue we haven’t shown that binds this configuration to a particular set of target hosts and that tells Ansible to make the changes as root. When we activate that binding (by running `ansible-playbook example.yml`), Ansible starts working to configure several target hosts in parallel. If any operation fails, Ansible reports the error and stops working on the host that generated it. Other hosts can continue until they’re done.
Client setup

Ansible needs three things from each configuration management client:

- SSH access
- **sudo** permission
- A Python 2 interpreter

Ansible does not actually require **sudo** access per se. It’s only needed if you want to run privileged operations. But you typically will.

If the client is a Linux cloud server, it may be Ansible-accessible right out of the box. Systems like FreeBSD that don’t install **sudo** or Python by default might need a bit more tweaking, but you can do some of the initial bootstrapping through Ansible with raw operations, which execute commands remotely without the usual Python wrapper. Or you can just write your own bootstrapping script.

Several choices must be made when Ansible clients are set up. We suggest a reasonable game plan in [Ansible access options](https://docs.ansible.com/#ansible-access-options), but for now, let’s assume you’ve created a dedicated “ansible” user on the client, that the appropriate SSH key is in your default set, and that you’re willing to enter the **sudo** password by hand.

Clients don’t introduce themselves to Ansible, so you need to add them to Ansible’s host inventory. By default, the inventory is a single file called `/etc/ansible/hosts`.

One nice feature of Ansible is that you can replace any flat configuration file with a directory of the same name. Ansible then merges the contents of the files the directory contains. This feature is useful for structuring your configuration base, but it’s also Ansible’s way of incorporating dynamic information: if a particular file is executable, Ansible runs it and captures the output instead of reading the file directly. Actually, Ansible is even smarter than this. It ignores certain file types entirely, e.g., `.ini` files. So not only can you put in scripts, but also configuration files for scripts.

This aggregation feature is so useful and so commonly used that we recommend bypassing the larval flat-file stage of most configuration files and skipping directly to directories. For example, we can define an Ansible client by adding the following line to `/etc/ansible/hosts/static` (or to `~/.hosts/static` within a personal configuration base):

```bash
new-client.example.com ansible_user=ansible
```

FreeBSD clients put Python in an unusual location, so you’ll need to tell Ansible about that:

```bash
freebsd.example.com ansible_python_interpreter=/usr/local/bin/python
ansible_user=ansible
```
This should all be on a single line in the **hosts** file. (On this page we present a much better way to set these variables, but that method is just a generalization of this same idea.)

To check connectivity with a new host, run the **setup** operation, which returns the client’s fact catalog:

```bash
$ ansible new-client.example.com -m setup
new-client.example.com | SUCCESS => {
  "ansible_facts": {
    "ansible_all_ipv4_addresses": [
      "172.31.25.123"
    ],
    ...
  <200+ more lines omitted>
```

The name “setup” is unfortunate, as no explicit setup is actually required. You can go directly to actual configuration operations if you wish. In addition, you can run the **setup** operation as often as you like to review the client’s fact catalog.

Check to be sure that privilege escalation through **sudo** is also working correctly:

```bash
$ ansible new-client.example.com -a whoami --become --ask-become-pass
SUDO password: <password>
new-client.example.com | SUCCESS | rc=0 >>
root
```

Here, the **command** operation, which runs shell commands, is the default. We could have said `-m command` explicitly with equivalent results. The `-a` flag introduces operation parameters; in this case, the actual command to execute.

“Becoming” is Ansible’s odd locution for privilege escalation; you “become” another user. The “other user” is root by default, but you can specify a different one with the `-u` option. Unfortunately, you have to force Ansible to ask you for the **sudo** password (with `--ask-become-pass`), and it does so regardless of whether the remote system actually prompts for the password.
Client groups

Groups are defined within the **hosts** directory as well, although the syntax can become a bit awkward:

```
client-four.example.com
[webservers]
  client-one.example.com
  client-two.example.com
[dbservers]
  client-one.example.com
  client-three.example.com
```

If this doesn’t look so bad, that’s because we’ve skirted the main problem areas. The **.ini** format is flat(ish), so some tricks are needed if you want to define hierarchical groups or add extras directly to the **hosts** file (e.g., variable assignments for a group). These features aren’t actually that important in practice, however.

Note that we had to list client-four at the top of the file because that host is not included in any groups. We can’t just append to the **hosts** file, because that would make client-four a member of the dbservers group, even if we added a blank line as a separator.

This is another reason why configuration directories are helpful. In practice, we’d probably want to put each group definition in a separate file.

Ansible lets you freely intermix client names and group names on command lines and within the configuration base. Neither is specially marked, and both can be subject to globbing. Regular-expression-style matching is also available for both; just start the pattern with a ~. There’s also a set algebra notation for combining cohorts of clients in various ways.

For example, the following command uses a globbing expression to select the webservers group. It executes the ping operation on each member of that group.

```
$ ansible 'web*' -m ping
```

```
client-one.example.com | SUCCESS => {
  "changed": false,
  "ping": "pong"
}
client-two.example.com | SUCCESS => {
  "changed": false,
  "ping": "pong"
}
```
Variable assignments

As we saw here, variable values can be assigned within inventory files. But that’s gauche; don’t do it that way.

Every host and group can have its own collection of variable definitions in YAML format. By default, these definitions are stored under /etc/ansible/host_vars and /etc/ansible/group_vars in files named for the host or group. You can use a .yml suffix if you want; Ansible finds the appropriate files either way.

As with other Ansible configurations, these files can be converted to directories if you’d like to add some additional structure or scripting. Ansible does its usual routine of ignoring configuration files, running scripts, and combining all the results into a final package.

Ansible automatically defines a group called “all” for you. Just like other groups, “all” can have its own group variables. For example, if you standardize on using client accounts named “ansible” for configuration management, that’s a good fact to put in the global configuration (here, in, say, group_vars/all/basics):

ansible_user: ansible

If multiple value declarations exist for a variable, Ansible selects a final value according to precedence rules rather than declaration order. Ansible currently has 14 different precedence categories, but the relevant point in this case is that host variables trump group variables.

Conflicts among overlapping groups are resolved at random, which can make for inconsistent behavior and tricky debugging. Try to structure your variable declarations so that there’s no possibility for overlaps.
**Dynamic and computed client groups**

Ansible’s grouping system really comes into its own when dynamic scripting is added to the mix. The dynamic inventory scripts used with cloud providers, for example, don’t simply list all the available servers. They also slice and dice those servers into ad hoc groupings according to metadata from the cloud.

For example, Amazon’s EC2 lets you assign arbitrary tags to each instance. You might assign the tag `webserver` to every instance that needs an NGINX stack and `dbserver` to every instance that needs PostgreSQL. The `ec2.py` dynamic inventory script would then create groups named `tag_webserver` and `tag_dbserver`. These groups can have their own group variables and can be named in bindings (“playbooks”), just like any other group.

The situation gets murkier when it comes to grouping clients on criteria internal to Ansible, such as the values of facts. You can’t do this directly. What you can do instead is target playbooks to broader groups (such as “all”) and apply conditional expressions to individual operations which, when the proper conditions do not apply, cause the operation to be skipped.

For example, the following playbook ensures that `/etc/rc.conf` contains a line to configure the hostname on each FreeBSD client:

```
- name: Set hostname at startup on FreeBSD systems
  hosts: all
  tasks:
    - lineinfile:
      dest: /etc/rc.conf
      line: hostname="{{ hostname }}"
      regexp: ^hostname
      when: ansible_os_family == "FreeBSD"
```

(If the last line looks like it needs some `{% %}`, your instinct is good. This is actually just a bit of Ansible syntactic sugar to help keep configurations tidy. `when` clauses are always going to be Jinja expressions, so Ansible surrounds their contents with double braces for you automatically. This feature is helpful, but it’s just one of a fairly extensive list of irregularities in Ansible’s YAML parsing.)

In this example, every host in inventory is considered for the `lineinfile` operation. But thanks to the `when` clause, only FreeBSD hosts actually run it. This approach works fine, but it doesn’t make the FreeBSD hosts into a true group. They can’t, for example, have a normal `group_vars` entry, although you can simulate the effect with some jury-rigging.

A structurally preferable but slightly more verbose alternative is to use a `group_by` operation, which runs locally and classifies hosts according to an arbitrary key value for which you designate a template:
The basic game plan is similar, but the classification occurs in a separate “play” (Ansible’s term for what we call a binding; see this page). We then start a new play so that we can specify a different set of target hosts, this time using the FreeBSD group that the first play defined for us.

The advantage of using `group_by` is that we perform the classification only once. We can then hang any number of tasks off the second play with confidence that we’re targeting only the intended clients.
Task lists

Ansible calls operations “tasks,” and a collection of tasks in a separate file is called a task list. Like all but a few parts of an Ansible configuration, task lists are just YAML, so the files have a .yml suffix.

The binding of task lists to specific hosts is done in higher-level objects called playbooks, which are described [here](#). For now, let’s focus on the operations themselves and not worry about how they come to be applied to a particular host.

As an example, we revisit the “install **sudo**” example from [this page](#) with a slightly different focus and implementation. This time, we create the administrator accounts from scratch and give each one its own UNIX group of the same name. We then set up a **sudoers** file that lists the administrators explicitly (instead of just assigning privileges to a “sudo” UNIX group).

Some input data is needed to drive these operations: in particular, the location of the **sudoers** file and the names and usernames of administrators. We should put this information in a separate variable file, say, **group_vars/all/admins.yml**:

```yaml
sudoers_path: /etc/sudoers
admins:
  - { username: manny, fullname: Manny Calavera }
  - { username: moe, fullname: Moe Money }
```

The value of `admins` is an array of hashes; we iterate through this array to create all the accounts. Here’s what the complete task list would look like:

```yaml
- name: Install sudo package
  package: name=sudo state=present

- name: Create personal groups for admins
  group: name={{ item.username }}
  with_items: "{{ admins }}"

- name: Create admin accounts
  user:
    name: "{{ item.username }}"
    comment: "{{ item.fullname }}"
    group: "{{ item.username }}"
    groups: wheel
    with_items: "{{ admins }}"

- name: Install sudoers file
  template:
    dest: "{{ sudoers_path }}"
    src: templates/sudoers.j2
    owner: root
    group: wheel
    mode: 0600
```
From the perspective of YAML and JSON, the tasks form a list. Each dash at the left margin starts a new task, which is represented by a hash.

In this example, every task has a name field that describes its function in English. The names are technically optional, but if you don’t include them, Ansible tells you very little about what it’s doing when you run the configuration (other than listing the module names: package, group, etc.).

Each task must have among its keys the name of exactly one operation module. The value of that key is itself a hash that enumerates the operation parameters. Parameters that you do not explicitly set assume default values.

The notation

- name: Install sudo package
  package: name=sudo state=present

is an Ansible extension to YAML that’s essentially equivalent to

- name: Install sudo package
  package:
    name: sudo
    state: present

There’s some potential weirdness here in the case of operations like shell that have “freeform” arguments, but we won’t rehash that here. See the YAML rant here.

The one-line format is not only more compact, but it also lets you set parameters whose values are Jinja expressions without quotes, as seen in the task that creates personal groups for admins. In the normal syntax, a Jinja expression cannot appear at the start of a value unless the entire value is in quotes. The quoting is benign, but it does add visual noise. Despite appearances, the quotes do not force the value to be a string.

Now we’re ready to break out a few of the more notable aspects of this example task list in the sections below.
**state parameters**

In Ansible, operation modules can often perform several different tasks depending on the state you request. For the package module, for example, state=present installs the package, state=absent removes it, and state=latest ensures that the package is both present and up to date. Operations often look for different sets of parameters depending on the state being invoked.

In a few cases (e.g., the service module with state=restarted, which restarts a daemon), this model wanders a bit from what might normally be conceived of as a “state,” but overall it works well. The state can be omitted (as shown here when creating the sudo group), in which case it assumes a default value, usually something positive and empowering such as present, configured, or running.
Iteration

`with_items` is an iteration construct that repeats a task once for each element it’s supplied with. For quick reference, here’s another copy of the two tasks in our example that use `with_items`:

- name: Create personal groups for admins
  group: name="{{ item.username }}"
  with_items: "{{ admins }}"

- name: Create admin accounts
  user:
    name: "{{ item.username }}"
    comment: "{{ item.fullname }}"
    group: "{{ item.username }}"
    groups: wheel
  with_items: "{{ admins }}"

Note that `with_items` is an attribute of the task, not the operation that the task runs.

On each pass through a loop, Ansible sets the value of `item` to one of the items supplied to `with_items`. In this case, we assigned the variable `admins` a list of hashes, so `item` is always a hash. The notation `item.username` is shorthand for `item['username']`. Use whichever you prefer.

Each of these tasks loops through the `admins` array separately. One pass creates UNIX groups and the other creates user accounts. Although Ansible does define a grouping mechanism for tasks (called a block), that construct unfortunately does not support `with_items`.

If you really need the effect of a single loop that executes multiple tasks in sequence, you can achieve it by moving the loop body into a separate file and including it into the main task list:

- include: sudo-subtasks.yml
  with_items: "{{ admins }}"

`with_items` is not the only loop available in Ansible. There are also loop forms dedicated to iterating over hashes (termed “dictionaries” in Python), over lists of files, and over globbing patterns.
Interaction with Jinja

The Ansible documentation is not very specific about how YAML and Jinja interact, but it’s important to understand the details. As constructs like `with_items` demonstrate, Jinja is not simply a preprocessor that’s run over a file before it is handed off to YAML (as is the case in Salt). In fact, Ansible parses YAML with Jinja expressions intact. It then Jinja-expands each string value immediately before use. Parameters of iterated operations are reevaluated during each iteration.

Jinja has control structures of its own, including loops and conditionals. However, they are inherently incompatible with Ansible’s delayed-evaluation architecture, and so they are not allowed in Ansible’s YAML files (although they can be used in templates). Ansible constructs such as `when` and `with_items` are not just window dressing for the equivalent Jinja. They represent a rather different approach to structuring the configuration.
Template rendering

Ansible uses the Jinja2 template language both to add dynamic features to YAML files and to flesh out configuration file templates installed by the template module. We use a template in this example to set up the sudoers file. Here are the variable definitions again for reference:

```
sudoers_path: /etc/sudoers
admins:
  - { username: manny, fullname: Manny Calavera }
  - { username: moe, fullname: Moe Money }
```

And the task code:

```
- name: Install sudoers file
template:
  dest: "{{ sudoers_path }}"
  src: templates/sudoers.j2
  owner: root
  group: wheel
  mode: 0600
```

The file sudoers.j2 is a mix of plain text and Jinja2 code for the dynamic bits. For example, here’s a skeletal example that gives “sudo ALL” privileges to each admin:

```
Defaults env_keep += "HOME"

{% for admin in admins %}
{{ admin }} ALL=(ALL) ALL
{% endfor %}
```

The for loop wrapped by {% %} is Jinja2 syntax. Unfortunately, you can’t indent loop bodies sensibly as you might in a real programming language, because doing so would cause the output of the template to be indented as well.

The expanded version looks like this:

```
Defaults env_keep += "HOME"
manny ALL=(ALL) ALL
moe ALL=(ALL) ALL
```

Note that variable values automatically flow through to templates. The values are available to configuration files under exactly the same names used to define them; no prefix or additional hierarchy is imposed. Autodiscovered fact variables are in the top-level namespace, too, but to forestall potential name conflicts they all begin with the prefix ansible_.

Ansible’s module for installing static files is called copy. However, you may as well treat all configuration files as templates, even if their contents initially consist of static text. You can then
add customizations in the future without having to touch the configuration code; just edit the template. Reserve copy for binary files and for static files that will never need expansion, such as public keys.
Bindings: plays and playbooks

Bindings are the mechanism through which tasks become associated with sets of client machines. Ansible’s binding object is called a play. Here’s a simple example:

```yaml
- name: Make sure NGINX is installed on web servers
  hosts: webservers
  tasks:
    - package: name=nginx state=present
```

Just as multiple tasks can be concatenated to form a task list, multiple plays in sequence form a “playbook.”

As in other systems, the basic elements of a binding are a set of hosts and a set of tasks. However, Ansible’s system allows several additional options to be specified at the play level. They’re listed in Table 23.3.

### Table 23.3: Ansible play elements

<table>
<thead>
<tr>
<th>Key</th>
<th>Format</th>
<th>What it specifies</th>
</tr>
</thead>
<tbody>
<tr>
<td>name</td>
<td>string</td>
<td>Name to print out when executing the play, optional</td>
</tr>
<tr>
<td>hosts</td>
<td>list, string</td>
<td>Client systems on which to run associated tasks and roles</td>
</tr>
<tr>
<td>vars</td>
<td>hash</td>
<td>Variable values to set for the scope of this play</td>
</tr>
<tr>
<td>vars_files</td>
<td>list</td>
<td>Files from which to read variable values</td>
</tr>
<tr>
<td>become*</td>
<td>strings</td>
<td>Privilege escalation (e.g., <code>sudo</code>) options</td>
</tr>
<tr>
<td>tags</td>
<td>list</td>
<td>Categories for selective execution</td>
</tr>
<tr>
<td>tasks</td>
<td>list</td>
<td>Operations to run; may include separate files</td>
</tr>
<tr>
<td>handlers</td>
<td>list</td>
<td>Operations to run in response to notify</td>
</tr>
<tr>
<td>roles</td>
<td>list</td>
<td>Bundles (roles) to invoke for these hosts</td>
</tr>
</tbody>
</table>

The biggies here are the variable-related options, not so much because they appear in plays per se, but because they’re available pretty much anywhere—even when executing `includes`. Ansible can activate the same task list or playbook again and again with different sets of variable values. It’s a lot like defining a function (e.g., “make a user account”) and then calling it with different sets of arguments.

Ansible formalizes this system in its implementation of bundles (called “roles”), which we discuss here. Roles are powerful, but under the hood, they’re just a set of standardized conventions for doing includes, so they’re also easy to understand.

Here’s a simple play that demonstrates the use of handlers:
This playbook runs on hosts clickera and clickerb. It mirrors files from a central (local) repository by running **rsync** (using the **synchronize** module), then restarts the NGINX web server if any updates were made.

When a task with a **notify** clause makes changes to the system, Ansible runs the handler of the requested name. Handlers themselves are just tasks, but they’re declared in a separate section of the play.

Playbooks are the primary unit of execution in Ansible. You run them with **ansible-playbook**:

```
$ ansible-playbook global.yml --ask-sudo-pass
```

Ansible approaches multihost execution task by task. As it reads a playbook, each task is run in parallel on the targeted hosts. When every host has completed the task, Ansible continues to the next task. By default, Ansible runs tasks simultaneously on up to five hosts, but you can set a different limit with the **-f** flag.

When debugging problems, it’s often helpful to include the **-vvvv** argument to increase the amount of debugging output. You’ll see the exact commands that are executed on the remote system and their detailed responses.
Roles

As we described generically starting here, bundles (our term) are the packaging mechanism defined by a CM system to facilitate reuse and sharing of configuration fragments.

Ansible calls its bundles “roles,” and they are in fact nothing but a structured system of include operations and variable precedence rules. They make it easy to put the variable definitions, task lists, and templates associated with a configuration into a single directory, making them readily available for reuse and sharing.

Each role is a subdirectory of a directory called roles that’s normally found at the top level of your configuration base. You can also add site-wide role directories by setting the roles_path variable in ansible.cfg, as shown here. All known role directories are searched whenever you include a role in a playbook.

Role directories can have the subdirectories shown in Table 23.4.

<table>
<thead>
<tr>
<th>Subdir</th>
<th>Contents</th>
</tr>
</thead>
<tbody>
<tr>
<td>defaults</td>
<td>Default values for variables (overridable)</td>
</tr>
<tr>
<td>vars</td>
<td>Variable definitions (not overridable, but can reference overrides)</td>
</tr>
<tr>
<td>tasks</td>
<td>Tasks lists (sets of operations)</td>
</tr>
<tr>
<td>handlers</td>
<td>Operations that respond to notifications</td>
</tr>
<tr>
<td>files</td>
<td>Data files (typically used as a source for copy operations)</td>
</tr>
<tr>
<td>templates</td>
<td>Templates to be processed by Jinja before installation</td>
</tr>
<tr>
<td>meta</td>
<td>List of bundles to run in preparation for this bundle</td>
</tr>
</tbody>
</table>

Roles are invoked through playbooks and nowhere else. Ansible looks for a file called main.yml within each of the role’s subdirectories. If it exists, the contents are automatically incorporated into any playbook that invokes the role. For example, the playbook

```yaml
- name: Set up cow-clicker app throughout East region
  hosts: web-servers-east
  roles:
    - cow-clicker
```

is roughly equivalent to
However, variable values from the `default` folder do not override values that have already been set. In addition, Ansible makes it easy to refer to files from the `files` and `templates` directories, and it sub-includes any roles mentioned as dependencies in the `meta/main.yml` file.

Files other than `main.yml` are ignored by the roles system, so you can break the configuration into whatever pieces are appropriate and just include those parts into `main.yml`.

Ansible lets you pass a set of variable values to a particular instance of a role. In effect, this makes the role act as a sort of parameterized function. For example, you might define a bundle that’s used to deploy a Rails app. You could invoke that bundle several times within a playbook, supplying the parameters of a different app for each invocation:

```yaml
- name: Install ULSAH Rails apps
  hosts: ulsah-server
  roles:
    - { role: rails_app, app_name: ulsah-reviews }
    - { role: rails_app, app_name: admin-com }
```

In this example, the `rails_app` role would probably depend on a role for `nginx` or some other web server, so it would not be necessary to mention the web server role explicitly. If you wanted to customize the web server installation, you could simply include the appropriate variable values in the `rails_app` invocation, and those values would be propagated downward.

Ansible’s public role repository is located at galaxy.ansible.com. You can search for roles with the `ansible-galaxy` command, but you’re better off using the web site. It lets you sort by rating or download count, and you can easily click through to the GitHub repo that hosts the actual code for each role. Several roles are usually available to address most common scenarios, so it’s worth examining the code to determine which version will serve your needs best.

Once you’ve settled on a role implementation, copy the files to your `roles` directory by running `ansible-galaxy install`. For example:

```
$ ansible-galaxy install ANXS.postgresql
- downloading role 'postgresql', owned by ANXS
- downloading role from https://github.com/ANXS/postgresql/v1.4.0.tar.gz
- extracting ANXS.postgresql to /etc/ansible/roles/ANXS.postgresql
- ANXS.postgresql was installed successfully
```
Recommendations for structuring the configuration base

Most configuration bases are organized hierarchically. That is, various pieces of the configuration feed into a master playbook that controls the global state. However, you can also define task-specific playbooks that are unrelated to the global scheme.

Try to keep task lists and handlers out of playbook files. Instead, put them in separate files and interpolate them with `include`. This structure makes a clean separation between bindings and behavior, and it puts all tasks on an equal footing. For extra style points, avoid freestanding task lists entirely and standardize on roles.

It’s sometimes recommended that a single playbook should cover all the tasks that relate to each logically distinct group of hosts. For example, all the roles and tasks that relate to web servers should be included in a single `webserver.yml` playbook. This approach avoids replication of host groups and provides a clear locus of control for each host group.

On the other hand, following this rule means that there’s no direct way to run a portion of the global configuration, even for debugging. Ansible can run only playbooks; there is no simple command that runs a specific task list on a given machine.

The official solution for this issue is tagging, which works fine but requires some setup. You can include a `tags` field in or above any task to classify it. At the command line, use `ansible-playbook`’s `-t` option to specify the subset of tags you want to run. In most debugging scenarios, you’ll also want to use the `-l` option to limit execution to a specified test host.

Assign tags at as high a level as you can within the configuration hierarchy. Under normal circumstances, you should have no temptation to assign tags to individual tasks. (If you do, it may be a sign that the particular task list should be split up.) Instead, attach tags to the `include` or `roles` clause that incorporates a specific task list or role into the configuration. The tags then cover all the included tasks.

Alternatively, you can just construct scratch playbooks that run parts of the configuration base on a test host. Setting up these scratch playbooks is a minor annoyance, but so is tagging.
Ansible access options

Ansible needs SSH and sudo access on every client system, which sounds straightforward and familiar until you consider that the configuration management system holds the master keys to the entire organization. It’s hard for daemon-based systems to be more secure than the root account on the configuration server, but Ansible can potentially do better than this with some thoughtful planning.

For simplicity, it’s best if SSH access is funneled through a dedicated account such as “ansible” that has the same name on each client. That account should use a simple shell and should have a minimal dot-file configuration.

On cloud servers, you can use a standard bootstrapping account (such as ec2-user on EC2) for Ansible control. Just make sure that after the initial setup, the account has been properly locked down and does not allow, e.g., su to root without a password.

You have some flexibility regarding the actual security design. But keep the following points in mind:

- Ansible needs one credential (password or private key) to gain access to a remote system, and another to escalate privileges with sudo. Proper security hygiene suggests that these be separate credentials. A single compromised credential should not grant an intruder root access to a target machine.

- If both credentials are stored in the same place with the same form of protection (encryption, file permissions), they are effectively a single credential.

- Credentials can be reused on machines that are peers (e.g., web servers in a farm), but it should not be possible to use credentials from one server to access a more sensitive—or even substantially different—server.

- Ansible has transparent support for encrypted data through the ansible-vault command, but only if the data is contained in a YAML or .ini file.

- Administrators can remember only a few passwords.

- It’s unreasonable to demand more than one password for a given operation.

Some sites set up client-side “ansible” accounts with the NOPASSWD option in the sudoers file, such that no password is required for the ansible account to run sudo. This is a terribly insecure configuration. If you can’t bring yourself to type a password, at least install the PAM SSH agent module and require a forwarded SSH key for sudo access. See this page for more information about PAM.

Sites will arrive at their own tradeoffs, but we suggest the following system as a robust but usable baseline that conforms to these guidelines:
• SSH access is controlled by key pairs that are used by Ansible only.

• Password-based SSH access is prohibited on client systems (by setting PasswordAuthentication no in /etc/ssh/sshd_config).

• SSH private keys are protected by a passphrase (set with ssh-keygen -p). All private keys have the same passphrase.

• Private SSH keys are kept in a known location on the Ansible master machine. They do not live within the configuration base, and administrators agree not to copy them elsewhere.

• Remote accounts (“ansible” accounts) have random UNIX passwords that are listed in the configuration base in encrypted form. All of them are encrypted with the same passphrase, but it’s different from the passphrase used for SSH private keys. You will need to add some Ansible glue to make sure the right passwords are used with the right client hosts.

In this scheme, both sets of credentials are encrypted, which makes them resistant to simple violations of file permissions. This layer of indirection also lets you change the master passphrases easily without changing the underlying keys.

Administrators need remember only two passphrases: the passphrase that gives access to SSH private keys, and the Ansible vault password, which allows Ansible to decrypt the host-specific sudo-passwords (as well as any other confidential information included in your configuration base).

If you require more granularity for administrator permissions (which is likely), you can encrypt multiple sets of credentials with different passphrases. If the sets are cumulative (as opposed to disjoint), no individual administrator needs to remember more than two passphrases.

See this page for more details on ssh-agent.

It’s assumed in this system that administrators will use ssh-agent to manage access to private keys. All keys can be activated with a single ssh-add command, and the SSH password need be entered only once per session. To work on a system other than the usual Ansible master, admins can use SSH’s ForwardAgent option to tunnel keys through to the machine on which work is being done. All other security information is included in the configuration base itself.

It’s true that ssh-agent and key forwarding are only as secure as the machines on which they run. (Less so, really: like sudo with a grace period, they are only as secure as your personal account.) However, the risk is mitigated by limits on time and context. Use the -t argument to ssh-agent or ssh-add to cap the lifetime of activated keys, and terminate connections that have access to forwarded keys once you are no longer using them.
If possible, private keys should never be deployed onto client systems. If clients need privileged access to controlled resources (e.g., to clone a controlled Git repository), use the proxying features built into SSH and Ansible, or use ssh-agent to make private keys temporarily available to the client without copying them.

For some reason, Ansible cannot currently recognize encrypted files in the configuration base and prompt you to enter the passphrase for decryption. You have to force its hand with the --ask-vault-pass argument to the ansible-playbook and ansible commands. There’s a --vault-password-file option available for noninteractive use, but of course, that reduces security. If you decide to use a password file, it should be accessible only to the dedicated ansible account.
23.6 **INTRODUCTION TO SALT**

Out in the world, you might see Salt referred to as Salt, SaltStack, or Salt Open. These terms are essentially interchangeable. The vendor’s name is SaltStack, and they use SaltStack as a generic term to refer to the complete product line, which includes some enterprise add-ons that we don’t discuss in this book. However, many people call the open source system SaltStack, too.

Salt Open is a more recently introduced name that designates only the open source components of Salt. But currently, that name doesn’t seem to be used anywhere outside of saltstack.com.

SaltStack maintains its own package repository at repo.saltstack.com which hosts up-to-date packages for every Linux packaging system. See the web site for instructions on how to add the repo to your configuration. Some distributions include free-range Salt packages of their own, but it’s generally best to go directly to the source.

You’ll need the **salt-master** package on the configuration server (the “master”). If you have any dealings with cloud providers, also install the **salt-cloud** package. It wraps a variety of cloud providers into a standard interface and simplifies the process of creating new cloud servers to be managed through Salt. It’s essentially similar to cloud providers’ native CLI tools, but it handles machines at both the Salt and cloud layers. New machines are automatically bootstrapped, enrolled, and approved. Deleted machines are removed from Salt as well as the provider’s cloud.

SaltStack doesn’t host a package repo for FreeBSD, but it is a supported platform. The web installer is FreeBSD aware:

```
$ curl -L https://bootstrap.saltstack.com -o /tmp/saltboot
$ sudo sh /tmp/saltboot -P -M
```

By default, the web installer installs client-side software as well as the master server. If you don’t want that, pass the `-N` option to **saltboot**.

---

*See Chapter 25 for more information about containers.*

---

Salt’s configuration files go in `/etc/salt`, both on the master server and on clients (“minions”). It’s theoretically possible to run the server daemon as an unprivileged user, but that requires manually `chowning` a bunch of system directories that Salt expects to interact with. If you’re tempted to head down this road, you’re probably better off using a containerized version of the server or saving the configuration into a pre-baked machine image.

Salt has a simple access control system that you can configure to allow unprivileged users to initiate Salt operations on minions. However, you must do manual permission hacking similar to that required for nonroot operation. Considering that the master has direct root access to all
minions, we find this feature rather suspect from a security perspective. If you do use it, keep a tight lid on the permissions that are granted.

Salt maintains a separation between configuration files that set variable values (the “pillar”) and configuration files that define operations (“states”). The distinction goes all the way to the top: you must set separate locations for these configuration hierarchies. They both default to living under /srv, which is equivalent to the following /etc/salt/master file:

```yaml
file_roots:
  base:
    - /srv/salt
pillar_roots:
  base:
    - /srv/pillar
```

Here, base is a required common environment on top of which additional environments (e.g., development) can be layered. Variable definitions go in the /srv/pillar root, and everything else lives in /srv/salt.

Note that the paths themselves are list elements, since they’re prefixed with dashes. You can include multiple directories, which makes the salt-master daemon serve a merged view of the listed directories to minions. This is a useful feature when you are organizing a large configuration base, since it permits you to add structure that Salt wouldn’t natively understand.

Typically, you’ll want to manage the configuration base as a single Git repository that includes both the salt and pillar subdirectories. This isn’t a good fit for the default layout because it means that /srv would be the repo root; consider moving everything down a level to /srv/salt/salt and /srv/salt/pillar.

The Salt documentation doesn’t do a very good job of explaining why the pillar and the states have to be completely separate, but in fact this distinction is central to Salt’s architecture. The salt-master daemon doesn’t pay the slightest attention to state files; it simply makes them available to minions, who are responsible for parsing and executing them.

The pillar is entirely different. It’s evaluated on the master and propagated to minions as a single, unified JSON hierarchy. Each minion sees a different view of the pillar, but none of them can see the implementation machinery behind these views.

In part, this is a security measure: Salt makes a strong guarantee that minions cannot access each other’s pillars. It’s also a data-sourcing distinction, as dynamic pillar content always originates from the master. This makes for a nice complementarity with grains (Salt’s version of facts), which originate on minions.

See this page for more information about network firewalls.
Salt’s communication bus uses TCP ports 4505 and 4506 on the server. Make sure these ports are allowed through any firewalls or packet filters that lie between the server and the prospective clients. The clients themselves do not accept network connections, so this step needs to be done only once, for the server.

When first investigating Salt, you might find it informative to run `salt-master -l debug` in a terminal window (instead of as a system service). This makes `salt-master` run in the foreground and print out activity on Salt’s communication bus as it occurs.
Minion setup

As on the master side, you have a choice of native packages from SaltStack’s repo or a universal bootstrap script. The repo is hardly worth fussing with on minions, so we recommend the latter:

```
$ curl -o /tmp/saltboot -sl https://bootstrap.saltstack.com
$ sudo sh /tmp/saltboot -P
```

The bootstrap script works on any supported system. On systems without curl, wget and fetch also work fine. See the saltstack/salt-bootstrap repository on GitHub for specific installation scenarios and source code.

For production systems that are started automatically, minimize your exposure to external events by downloading a locally cached version of the boot script. Install a specific version of the Salt client, also from a local cache, or preload it on the machine image. Run the boot script with a `-h` option to see all the options it supports.

---

See [Chapter 16](#) for more information about DNS.

---

By default, the salt-minion daemon tries to register itself with a master machine named “salt”. (This “magic name” system was first popularized by Puppet.) You can use DNS wizardry to make the name resolve appropriately, or you can set an explicit master in `/etc/salt/minion` (/usr/local/etc/salt/minion on FreeBSD):

```
master: salt.example.com
```

Restart salt-minion after modifying this file (usually, `service salt_minion restart`, note the underscore rather than a dash).

salt-master accepts client registrations from any random machine that can reach it, but you must approve each client with the salt-key command on the master configuration server before it becomes active:

```
$ sudo salt-key -l unaccepted
Unaccepted Keys:
new-client.example.com

# If everything looks good, accept all pending keys

$ sudo salt-key -yA
The following keys are going to be accepted:
Unaccepted Keys:
new-client.example.com
Key for minion new-client.example.com accepted.
```

You can now check connectivity from the server with the test module:
In this example, new-client.example.com looks suspiciously like a hostname, but it really isn’t. It’s just the machine’s Salt ID, a string that defaults to the hostname but can be set to anything you like in the client’s /etc/salt/minion file:

```
master: salt.example.com
id: new-client.example.com
```

IDs and IP addresses have nothing to do with each other. For example, even if 52.24.149.191 were the client’s actual IP address, you could not directly target it that way with Salt commands:

```
$ sudo salt 52.24.149.191 test.ping
No minions matched the target. No command was sent, no jid was assigned.
ERROR: No return received
```

(Of course, you can do IP-based matching. It just has to be explicit. See this page.)
Variable value binding for minions

As we saw in the server setup section, Salt has separate filesystem hierarchies for state bindings and variable-value bindings (the “pillar”). Each of these directory trees has a top.sls file at the root that binds groups of minions to files within the tree. The two top.sls files both use the same layout. (.sls is just Salt’s standard extension for YAML files.)

As an example, here’s the layout of a simple Salt configuration base that shows both the salt and pillar roots:

```
$ tree /srv/salt
/srv/salt
    |--- salt
    |     |--- top.sls
    |     |--- hostname.sls
    |     |--- bootstrap.sls
    |     |--- sshd.sls
    |     |--- baseline.sls
    |--- pillar
    |     |--- top.sls
    |     |--- baseline.sls
    |     |--- webserver.sls
    |     |--- freebsd.sls

2 directories, 9 files
```

To bind the variables defined in pillar/baseline.sls and pillar/freebsd.sls to our example client, we could include the following lines in pillar/top.sls:

```
base:
    new-client.example.com:
    - baseline
    - freebsd
```

As in the master file, base is a required, common environment that can be overlaid in more sophisticated setups. See this page for more about this.

It’s possible for baseline.sls and freebsd.sls to define some of the same variable values. For scalar and array values, the last source listed in top.sls is the one that takes effect. Hashes, however, are merged.

For example, if a minion binds to one variable file that looks like this:

```
admin-users:
    manny:
        uid: 724
    moe:
        uid: 740
```
and one that looks like this:

```yaml
admin-users:
  jack:
    uid: 1004
```

then Salt merges the two versions.

The pillar data presented to minions is

```yaml
admin-users:
  manny:
    uid: 724
  moe:
    uid: 740
  jack:
    uid: 1004
```
Minion matching

In the scenario above, what we probably want is to apply `baseline.sls` to all clients, and to apply `freebsd.sls` to all clients that are running FreeBSD. Here’s how we can do that with selection patterns in the `pillar/top.sls` file:

```yaml
base:
  '*.*.example.com':
    - baseline
  'G@os:FreeBSD':
    - freebsd
```

The star matches all client IDs in example.com. We could have just used `*` here, but we wanted to emphasize that it’s a globbing pattern. The `G@` prefix requests a match on grain values. The grain being inspected is named `os`, and the value sought is `FreeBSD`. Globbing is allowed here, too.

A less magical way to write the matching expression for FreeBSD would be

```yaml
'os:FreeBSD':
  - match: grain
  - freebsd
```

The choice is up to you, but the `@` notation expands cleanly to complex expressions that involve parentheses and Boolean operations. Table 23.5 lists most of the common matching types, although a few have been omitted.

**Table 23.5: Salt minion match types**

<table>
<thead>
<tr>
<th>Code</th>
<th>Target</th>
<th>Match type</th>
<th>match:</th>
<th>Example</th>
</tr>
</thead>
<tbody>
<tr>
<td>–</td>
<td>ID</td>
<td>glob</td>
<td>glob</td>
<td><em>.</em>.cloud.example.com</td>
</tr>
<tr>
<td>E</td>
<td>ID</td>
<td>regex</td>
<td>pcre</td>
<td>E@(nw</td>
</tr>
<tr>
<td>L</td>
<td>ID</td>
<td>list</td>
<td>list</td>
<td>L@hosta,hostb,hostc</td>
</tr>
<tr>
<td>G</td>
<td>grain</td>
<td>glob</td>
<td>grain</td>
<td>G@domain:.*.example.com</td>
</tr>
<tr>
<td>E</td>
<td>grain</td>
<td>regex</td>
<td>grain_pcre</td>
<td>E@virtual:(xen</td>
</tr>
<tr>
<td>I</td>
<td>pillar</td>
<td>glob</td>
<td>pillar</td>
<td>I@scaling_type:autoscale</td>
</tr>
<tr>
<td>J</td>
<td>pillar</td>
<td>regex</td>
<td>pillar_pcre</td>
<td>J@server-class:(web</td>
</tr>
<tr>
<td>S</td>
<td>IP address</td>
<td>CIDR block</td>
<td>ipcidr</td>
<td>S@52.24.9/20</td>
</tr>
<tr>
<td>–</td>
<td>compound</td>
<td>compound</td>
<td>compound</td>
<td>not G@os_family:RedHat</td>
</tr>
</tbody>
</table>

a. This is the default. No match-type code is needed (or defined).
b. Note the lack of spaces; individual expressions can’t include them.
c. Codes are used to label individual terms.

If Table 23.5 looks disturbingly complex, take heart; these are just options. Real-world selectors look a lot more like our simple examples.
If you’re wondering what all those grains or pillar values are that you can match against, it’s easy to find out. Just use

```bash
$ sudo salt minion grains.items
```

or

```bash
$ sudo salt minion pillar.items
```

to obtain a complete list.

You can define named groups in the `/etc/salt/master` file. They’re called nodegroups, and they are useful for moving complex group selectors out of `top.sls` files. However, they’re not really a true grouping mechanism so much as a way to name patterns for reuse. As a result, their behavior is a bit squirrely. They can only be defined in terms of compound-type selectors (not, for example, by a simple list of clients, unless you use an `L@` clause), and you must use an explicit `match` statement: `type of nodegroup` to match against them. There’s no global shorthand notation.
Salt states

Salt operations are called “states.” As in Ansible, they’re defined in YAML format, and in fact they look vaguely similar to Ansible tasks. However, the fine-grained details are quite different. You can include a series of state definitions in a .sls file.

States are bound to specific minions in the top.sls file at the root of the salt arm of the configuration base. This file looks and functions exactly like the top.sls file for variable bindings; see the examples here.

Take a look at the following Salt version of the same example we worked through with Ansible starting on this page: we install sudo and create a corresponding sudo group to which we assign administrators who should have sudo privileges. We then create a group of administrator accounts, each of which has its own UNIX group of the same name. Finally, we then copy in a sudoers file from the configuration base.

As it happens, we can use exactly the same variable file for Salt that we used for Ansible:

```yaml
sudoers_path: /etc/sudoers
admins:
    - { username: manny, fullname: Manny Calavera }
    - { username: moe, fullname: Moe Money }
```

To make these definitions available to all minions, we put them in the configuration base at pillar/example.sls and add a binding to top.sls:

```yaml
base:
    '*':
        - example
```

Here’s a Salt version of the operations:
install-sudo-package:
  pkg.installed:
    - name: sudo
    - refresh: true

create-sudo-group:
  group.present:
    - name: sudo

{% for admin in pillaradmins %}
create-group-{{ admin.username }}:
  group.present:
    - name: {{ admin.username }}
create-user-{{ admin.username }}:
  user.present:
    - name: {{ admin.username }}
    - gid: {{ admin.username }}
    - groups: [wheel, sudo]
    - fullname: {{ admin.fullname }}
{% endfor %}
  - mode: '0600'
  - user: root
    - name: {{ pillar.sudoers_path }}
    - group: wheel
install-sudoers-file:
  - source: salt://files/sudoers
file.managed:

This version shows the operations in their most canonical form for easier comparison with the equivalent Ansible task list that starts here. We can make a few additional changes to clean things up a bit, but first, a look at this longer version.
Salt and Jinja

The first thing to notice is that the file includes a Jinja loop delimited by { and %}. These delimiters are similar to {{ and }} except that { and %} do not return values. The contents of the loop are interpolated into the YAML file as many times as the loop runs.

See this page for general information about Python.

Although Jinja uses Python-like syntax, YAML already “owns” the indentation in a .sls file, so Jinja is forced to define block-ending tokens such as endfor. In straight Python, blocks would normally be defined through indentation.

Salt defines only a rudimentary iteration construct in its basic YAML scheme (see the comments regarding names here). Conditionals and robust iteration have to be provided by Jinja, or by whatever template language the .sls file is run through. (In fact, Salt does not care about YAML, either. It just expands configuration files through a designated pipeline and consumes the final JSON output, which must be fully literal.)

On one hand, this approach is clean. There’s no conceptual ambiguity about what’s going on, and it's easy to examine an expanded .sls file to make sure it means what you intended. On the other hand, it means you’ll be using Jinja to provide any logic required by your configuration. The mix of templating code and YAML can easily become somewhat dazzling. It’s a bit like writing the logic of a web app using only HTML templates.

Several rules of thumb can help keep Salt configurations tidy. First, Salt has usable and well-defined mechanisms for implementing variable-value overlays. Use these to keep as much configuration as possible in the domain of data rather than code.

Many examples in the Salt documentation use Jinja conditionals when they aren’t the best solution, for example. (In fairness, the examples are usually designed to illustrate some point other than general tidiness.) The following .sls file installs the Apache web server, which has different package names on different distributions:

```yaml
# apache-pkg.sls
apache:
    pkg.installed:
        {% if grains['os'] == 'RedHat' %}
            - name: httpd
        {% elif grains['os'] == 'Ubuntu' %}
            - name: apache2
        {% endif %}
```

This variation could be dealt with more elegantly through the pillar:
Although replacing one file with four might not initially seem like a simplification, it’s now an extensible and code-free system. Multi-OS environments will encounter many such variations, and they can all be dealt with in one place.

If a value has to be dynamically calculated, consider whether you can put the code at the top of the `.sls` file and simply memorialize it for later use in a variable. For example, another way to write the Apache package installation above would be

```bash
# pillar/top.sls
base:
  '*':
    - defaults
    'G@os:Ubuntu':
      - ubuntu

# pillar/defaults.sls
apache-pkg: httpd

# pillar/ubuntu.sls
apache-pkg: apache2
```

This at least has the advantage of separating the Jinja logic from the actual configuration.

If you must intermix Jinja logic with YAML, consider whether you can break out some of the YAML segments into separate files. You can then interpolate these segments as appropriate. Once again, the idea is simply to separate the code and YAML rather than alternating back and forth between them.

For nontrivial calculations, you can abandon YAML altogether and replace it with pure Python, or with one of the Python-based DSLs that Salt includes by default. See the Salt documentation for “renderers” for more information.
State IDs and dependencies

To return to our `sudo` example from this page, here are its first two states again for reference:

```yaml
install-sudo-package:
  pkg.installed:
    - name: sudo
    - refresh: true
create-sudo-group:
  group.present:
    - name: sudo
```

You can see that the individual states are not items in a list (as they are in Ansible) but rather the elements of a hash. The hash key for each state is an arbitrary string called the ID. As usual with hashes, IDs must be unique or they’ll collide.

But wait! The potential domain for collisions is not just this particular file, but the entire client configuration. State IDs must be globally unique, because Salt is eventually going to stuff them all together into one big hash.

It’s a bit of a funny hash, though, because it preserves the order of keys. In a standard hash, keys emerge in random order when the hash is enumerated. That’s the way that Salt used to work, too, and as a result, all dependencies among states had to be explicitly declared. These days, the hash preserves the order of presentation by default, although that can still be overridden if explicit dependencies are declared (or if this behavior is turned off in the `master` file).

There’s still some trickiness, though. In the absence of other constraints, order of execution conforms to the original `.sls` files. However, Salt still presumes that states are not logically dependent on one another unless you say so. If a state fails to execute, Salt notes the error but then continues and runs the next state.

If you want a dependent state not to run if its ancestors fail, you can declare that explicitly. For example:

```yaml
create-sudo-group:
  group.present:
    - name: sudo
    - require:
      - install-sudo-package
```

In this configuration, Salt won’t try to create a sudo group unless the `sudo` package was successfully installed.

Requisites also come into play when ordering states from multiple files. Unlike Ansible, Salt does not interpolate the contents of an `include` file at the point the `include` was encountered. It simply adds the file to its to-read list. If multiple files attempt to include the same source, there is still be
only one copy of the source in the final assembly, and the order of states might not be what you expected. In-order execution is guaranteed only within a file; if any states depend on externally defined operations, they must declare explicit requisites.

The requisite mechanism is also used to achieve an effect analogous to Ansible’s notifications. Actually, a handful of alternatives to `require` are syntactically interchangeable with it but imply subtle shadings of behavior. One of those, `watch`, is particularly useful for doing things when another state makes changes to the system.

For example, the following configuration sets the system’s time zone and the arguments to be passed to `ntpd` when it starts up. This configuration always makes sure that `ntpd` is running and configured to start at boot time. In addition, it restarts `ntpd` if either the system time zone or the `ntpd` flags are updated.

```yaml
set-timezone:
  timezone.system:
    - name: America/Los_Angeles

set-ntpd-opts:
  augeas.change:
    - context: /files/etc/rc.conf
    - lens: shellvars.lns
    - changes:
      - set ntpd_flags "-g"

ntpd:
  service.running:
    - enable: true
    - watch:
      - set-ntpd-opts
      - set-timezone
```

Augeas is a tool that understands many different file formats and facilitates automated changes.
State and execution functions

In a .sls file, the names that appear directly under state IDs are the operations those states should run. Some specific cases from our example scenario are pkg.installed and group.present.

These names include both a “module” part and a “function” part. Together, they are roughly analogous to an Ansible module name together with a state value. For example, Ansible uses a package module with state=present for installing packages, whereas Salt uses a dedicated pkg.installed function within the pkg module.

Salt makes a big whoop-de-do of distinguishing operations that do things to target systems (“execution functions”) from those that idempotently enforce a particular configuration (“state functions”). State functions usually call their associated execution functions when they need to make changes.

The general idea is that only state functions should be mentioned in .sls files, and only execution functions should appear on command lines. Salt primly enforces these rules, sometimes to confusing effect.

State and execution functions live in separate Python modules, but related modules usually share the same name. For example, there’s both a timezone state module and a timezone execution module. There can’t be any overlap in function names between the two modules, though, because that would create ambiguity. The end result is that to set the time zone from a .sls file, you must use timezone.system:

```python
set-timezone:
    timezone.system:
        - name: America/Los_Angeles
```

But to set a minion’s time zone from the command line, you use timezone.set_zone:

```
$ sudo salt minion timezone.set_zone America/Los_Angeles
```

If you get it wrong and need to consult the documentation, you’ll find the two halves of timezone in different sections of the manual. It’s also not always clear from behavior exactly which type of function is which. For example, git.config_set, which sets Git repository options, is a state function, but state.apply, which idempotently enforces configurations, is an execution function.

Ultimately, you just have to know which functions are which and the contexts to which they belong. If you need to call a function from the “wrong” context—which is sometimes necessary—you can use the adapter functions module.run (runs an execution function from a state context) and state.single (runs a state function from an execution context). For example, the adapted timezone calls above would be
set-timezone:
  module.run:
    - name: timezone.set zone
    - timezone: America/Los_Angeles

and

# salt minion state.single timezone.system name=America/Los_Angeles
**Parameters and names**

Once again, here are the first two states from [this page](#) for reference:

```yaml
install-sudo-package:
    pkg.installed:
        - name: sudo
        - refresh: true

create-sudo-group:
    group.present:
        - name: sudo
```

Indented under the name of each operation (that is, the `module.function` construction) is its list of parameters. In Ansible, the parameters for an operation form one big hash. Salt wants them as a list, with each entry prefaced by a dash. More specifically, Salt wants a list of hashes, though there’s typically only one key in each hash.

Most parameter lists include a parameter called `name`, which is the standard label for “the thing this operation is configuring.” Alternatively, you can supply a list of targets in a parameter called `names`. For example:

```yaml
create-groups:
    group.present:
        - names:
            - sudo
            - rvm
```

If you provide a `names` parameter, Salt reruns the operation multiple times, substituting one item from the `names` list into the `name` parameter on each pass. This is a mechanical process, and the operation itself is not aware of the iteration. It’s a run-time (as opposed to parse-time) operation, much like Ansible’s `with_items` construction. But because Jinja expansion has already completed, there’s no opportunity to base the values of other parameters on the `name`. If you need to adjust multiple parameters, ignore `names` and just iterate with a Jinja loop.

Some operations can handle multiple arguments at once. For example, `pkg.installed` can hand off multiple package names at once to the underlying OS package manager, which may be useful for efficiency or dependency resolution. Because Salt hides `names` iteration, such operations are forced to use a separate parameter name to enable bulk operations. For example, the states

```yaml
install-packages:
    pkg.installed:
        - names: [ sudo, curl ]
```

and
both install **sudo** and **curl**. The first version does it in two distinct operations, and the second does it in one.

We stress this seemingly minor point because it’s easy to make mistakes with **names**. Because it’s mechanical, **names** iterates even operations that pay no attention to the **name** parameter. On reviewing the Salt log, you’ll see that multiple executions have run successfully, but somehow the target system still doesn’t seem to be properly configured. So it’s helpful to understand exactly what’s going on.

If you don’t specify an explicit **name** for a state, Salt copies the state ID to this field. You can use this behavior to simplify state definitions a bit. For example,

```
create-sudo-group:
  group.present:
    - name: sudo
```

becomes

```
sudo:
  group.present
```

or even just

```
sudo: group.present
```

YAML doesn’t allow hash keys without values, so now that **group.present** no longer has any listed parameters, it has to become a simple string instead of a hash key with a parameter list as a value. That’s fine; Salt checks for this explicitly.

The shorthand style is usually clearer than the long form. A separate ID field can theoretically serve as a comment or an explanation, but most IDs seen in the wild simply restate behavior that is already obvious. If you want comments, add comments.

The shorthand form has a potential problem, though: since state IDs must be globally unique, short IDs named for common system entities are more vulnerable to ID collisions. Salt detects and reports conflicts, so this is really more an annoyance than a serious issue. But if you’re writing a Salt formula with the intention of reusing it in several configuration bases or you are planning to share it with the Salt community, stick with IDs that are less likely to clash.

Salt allows several operations to be included in a single state. Since the two operations above share a **name** field, we can combine them into a single state without having to state any explicit **name**S. However, there’s yet another YAML snare awaiting us:
sudo:
  pkg.installed:
  - refresh: true
  group.present: []

The value of the `sudo` key now has to be a hash; it can’t be a hash with the string `group.present` somehow tacked on. Accordingly, we now have to treat `group.present` as a hash key and provide an explicit parameter list as a value, even though that list is empty. That’s true even if we drop the `refresh` parameter from `pkg.installed`:

```bash
sudo:
  pkg.installed: []
  group.present: []
```

Just as we collapsed these two states, we can collapse our two states that do user account management. A more idiomatic version of the state list from this page is thus

```bash
sudo:
  pkg.installed: []
  group.present: []

{% for admin in pillaradmins %}
  {{ admin.username }}:
    group.present: []
    user.present:
      - gid: {{ admin.username }}
      - groups: [ wheel, sudo ]
      - fullname: {{ admin.fullname }}
  {% endfor %}

{{ pillar.sudoers_path }}:
  file.managed:
    - source: salt://files/sudoers
    - user: root
    - group: wheel
    - mode: '0600'
```
State binding to minions

As it happens, there’s not much more to say about Salt state bindings. They work exactly like pillar bindings. There’s a top.sls file at the root of the state hierarchy, and it maps minion groups to state files. Here’s a skeletal example:

```yaml
base:
  '*':
    - bootstrap
    - sitebase
  'G@os:Ubuntu':
    - ubuntu
  'G@webserver':
    - nginx
    - webapps
```

In this configuration, all hosts apply states from bootstrap.sls and sitebase.sls from the root of the state hierarchy. Ubuntu systems also run ubuntu.sls, and web servers (that is, minions that have a top-level webserver entry in their grains databases) run states to configure NGINX and local web apps.

Order in top.sls corresponds to the general order of execution on each minion. But as usual, explicit dependency information within states overrides the default order.
Highstates

Salt refers to the bindings in top.sls as a minion’s “highstate.” There’s a bit of potential terminological confusion in that Salt also uses “highstate” to mean “a parsed and assembled JSON tree of states,” which it then processes to form a “lowstate”—also a JSON tree—which is the low-level input to the execution engine.

You activate the highstate by telling the minion to run the state.apply function with no arguments:

$ sudo salt minion state.apply

The state.highstate function is equivalent to state.apply with no arguments. You’ll see both forms used.

Especially when debugging new state definitions, you might want a minion to run only a single state file. That’s easily accomplished with state.apply:

$ sudo salt minion state.apply statefile

Leave out the .sls suffix on the state file name; Salt will add it. Also keep in mind that the path to the state file has nothing to do with your current directory. It’s always interpreted relative to the state root as defined in the minion’s configuration file. This command does not redefine the minion’s highstate in any way; it simply runs the specified state file.

The salt command accepts a variety of flags for targeting different sorts of minion groups, but it’s easiest to just remember -C for “compound” and use one of the shorthands from Table 23.5.

For example, to highstate all Red Hat minions:

$ sudo salt -C os:RedHat state.highstate

The default match type is ID globbing, so the command

$ sudo salt '* state.highstate

is the command for “validate the entire site’s configuration.”

In keeping with Salt’s minion-centric execution model, all parallel executions begin simultaneously, and minions do not report back until they have completed execution. The salt command prints each minion’s results as soon as it receives them. There is no way to display incremental results while a state file is executing.

If you have lots of minions or a complex configuration base, the salt command’s default output can be quite a lot to look through because it reports on every operation considered by every minion. Add the option --state-output=mixed to reduce this output to one line for operations that succeed and cause no changes. The option --state-verbose=false suppresses output for no-change operations entirely, but salt still prints a header and summary for each minion.
Salt formulas

Salt calls its bundles “formulas” (well, “formula,” really). Like Ansible roles, they’re just a directory of files, although Salt formulas have an outer wrapper that includes some metadata and versioning information as well. In actual use, you just need the inner formula directory.

Formula directories go in one of the salt roots defined in the master file. If you want, you can create a root just for formulas. Formulas sometimes include example pillar data, but you’re responsible for installing that yourself.

Salt does nothing special to support formulas, except that if you name a directory in a top.sls file or include statement, Salt looks for an init.yml file within that directory and reads that. This convention provides a clear default path into the formula. Many formulas also include standalone states that you can reference by specifying both the directory and filename.

Nothing in Salt can be included in a configuration more than once, and that includes formulas. You can make multiple inclusion requests, but they’ll be coalesced. As a result, formulas cannot be instantiated multiple times in the way that Ansible roles can.

It doesn’t matter anyway, because Salt defines no way to pass parameters to a formula other than by putting variable values in the pillar. (Jinja expressions can set the values of variables, but those settings exist only within the context of the current file.)

To simulate the effect of invoking a formula repeatedly, you can supply pillar data in the form of a list or hash that the formula can iterate through on its own. However, the formula must be explicitly written with this structure in mind. You can’t impose it after the fact.

The central Salt repository for community-contributed formulas is currently just GitHub. Look for the username salt-formulas. Each formula is a separate project.
Environments

See this page for more information about environments.

Salt makes several gestures toward explicit support for environments (e.g., the separation of development, test, and production universes). Unfortunately, its environment facilities are somewhat peculiar, and they don’t map straightforwardly to the most common real-world use cases. It’s possible to get environments up and running with a little bit of determination and a tube of Jinja glue, but we find that in practice, many sites simply punt and run separate master servers for each environment instead. This jibes well with security and compliance standards that require separation of environments at the network layer.

As we saw back on this page, the `/etc/salt/master` file enumerates the various places where configuration information can be stored. It also associates an environment with each set of paths:

```yaml
file_roots:
    base:
        - /srv/salt

pillar_roots:
    base:
        - /srv/pillar
```

Here, `/srv/salt` and `/srv/pillar` are the state and pillar root directories for the default environment, called `base`. For simplicity, we have omitted mention of pillar data in the discussion below; environment management works the same way for both arms of the configuration base.

Sites with more than one environment will typically add an additional layer to the configuration directory hierarchy to represent that fact:

```yaml
file_roots:
    base:
        - /srv/base/salt
    development:
        - /srv/development/salt
    production:
        - /srv/production/salt
```

(Evidently, these example cowboys have no test environment. Don’t try this at home!)

An environment can list multiple root directories. If there’s more than one, the server transparently merges their contents. However, each environment performs a separate merge, and the final results remain segregated.
Inside top.sls files (the bindings that associate minions to particular states and pillar files), top-level keys are always environment names. So far, we’ve only seen examples that used the base environment, but of course any valid environment can go in this spot. For example:

```
base:
  '*':
    - global
development:
  '*-dev':
    - webserver
    - database
production:
  '*web*-prod':
    - webserver
  '*db*-prod':
    - database
```

The exact import of an environment’s appearance in a top.sls file depends on how you’ve configured Salt. In all cases, environments must already be defined in the master file; top files cannot create new environments. In addition, state files are required to originate from the environment context in which they are mentioned.

By default, Salt does not associate minions with any particular environment, and minions can receive state assignments from any or all of the environments in top.sls. In the snippet above, for example, all minions run the global.sls state from the base environment. Depending on their IDs, individual minions may also receive states from the production or development environments. (When you set a minion’s ID to match the development or production pattern, you are functionally associating it to the corresponding environment. However, Salt itself does not make an explicit association—at least, not in this configuration.)

The Salt documentation encourages this way of configuring environments, but we have some reservations. One potential issue is that minions end up as frankenservers that draw configuration elements from multiple environments. You can’t trace any given minion’s configuration back to one particular environment at one particular point in time, because every minion has multiple parents.

This distinction is important because a single base environment must be shared among all other environments. Which one should it be? The development version of the base environment? The production version? A completely separate and staged configuration base? Exactly when should you migrate the base environment to a new release?

There’s also some additional complexity lurking under the covers. Each environment is a full-fledged Salt configuration hierarchy, so it can, in theory, have its own top.sls file. Each of those top.sls files can, in theory, refer to multiple environments. When confronted with this situation, Salt tries to merge all the top files into one composite frankenconfiguration. (Merging occurs at the YAML level, though, so you’d better hope that multiple top files don’t try to assign states to
the same matching pattern within the same environment. If they do, some states will be silently
dispatched. Environments can demand the execution of one another’s states—states that they
don’t own, control, or know anything about. It would be horrifying if it weren’t so silly.

It’s not clear exactly what use cases this architecture attempts to enable. Although top file
merging is the default behavior, the docs repeatedly warn you away from setting things up this
way. Instead, you’re encouraged to designate a single `top.sls` file, most likely in `base`, to control
all environments.

If you do that, though, it soon becomes apparent that there’s some organizational friction
between this “external” top file and the rest of the environments. The top file is an integral part of
an environment’s configuration, so states and top files are normally co-developed; a change to
one often requires changes to the other. With a separate top file, you must effectively separate
each environment into two pieces that must be manually kept synchronized with each other. In
addition, the master top file must be shared with, synchronized with, and compatible with all
other environments. When you promote the test environment to production, for example, you
must make sure the master `top.sls` is adjusted to reflect the proper settings for that specific
version of the new production release.

Alternatively, you can hard-wire minions to a given environment, either by setting the value of
`environment` in the minion’s `/etc/salt/minion` file or by including the flag `saltenv=environment` on `salt`
command lines. Under this regime, a minion sees only the `top.sls` file of its assigned
environment. Within that top file, its view is also further limited to entries that appear under that
environment.

For example, a machine pinned to the development environment might see the `top.sls` file from
this page in the following abbreviated form (assuming that the `top.sls` file was found at the root
of the development state tree):

dev: development:
    '*-dev':
        - webserver
        - database

This mode of operation is quite a bit closer than the default to the traditional concept of
environments. There can be no unintended cross-talk among environments, which limits the
potential for unintended behavior. It also has the advantage that as a particular version of the
configuration base is promoted through the environment chain, different portions of the `top.sls`
file automatically apply themselves to clients.

The main disadvantage is that you lose the ability to factor out parts of the configuration that are
common to more than one environment. There’s no built-in way to “see” outside the context of
the current environment, so elements of the baseline configuration must be replicated into every
environment.
Rewritten to work in the context of this approach, the top.sls file from this page would look something like this:

```
development:
  '*':
    - global
  '*-dev':
    - webserver
    - database
production:
  '*':
    - global
  '*web*-prod':
    - webserver
  '*db*-prod':
    - database
```

The base environment itself is now vestigial, so we’ve dropped it from the top.sls file and copied that key’s former contents directly into the development and production environments.

Keep in mind that we’re now operating in a world where every environment tree has its own top.sls file. For this example, we assume that the top.sls file hasn’t diverged between the two environments, so the same contents would appear in both copies of top.sls.

Of course, manually reproducing the elements of the common configuration inside each environment is prone to error. A better option is to define the common configuration as a Jinja macro so that it can automatically be repeated:

```
{% macro baseline() %}
  '*':
    - global
{% endmacro %}

development:
{{ baseline() }}
  '*-dev':
    - webserver
    - database
production:
{{ baseline() }}
  '*web*-prod':
    - webserver
  '*db*-prod':
    - database
```

We’re assuming in this scenario that all minions are pinned to specific environments, so we can now potentially remove the environment indicators from minion IDs. However, it’s a good idea to retain them for security.
The issue is that minions control their own environment settings. If a minion in the development environment were compromised, for example, it could declare itself to be a production server and potentially gain access to the keys and configurations used in the production environment. (This is perhaps one reason why the Salt documentation seems a bit skittish about recommending environment pinning.)

Making environment-specific configurations contingent on both the environment settings and the minion IDs protects against this line of attack. If a minion changes its ID, the master no longer recognizes it as an approved client and ignores it until an administrator approves the change with the `salt-key` command.

If you prefer not to use IDs in this way, an alternative is to use pillar data as a cross-check. Whatever you do, you can’t just drop the suffix and turn ‘*-dev’ into ‘*’, because the shared portion of the configuration already uses ‘*’ as a key. Duplicate patterns within an environment are a YAML violation.

When debugging environments, you’ll find a couple of execution functions especially helpful. `config.get` shows the value that a particular minion (or set of minions) is using for a configuration option:

```
$ sudo salt new-client-dev config.get environment
new-client-dev:
  development
```

Here, we can see that the minion with ID new-client-dev has been pinned to the development environment, just as its ID would suggest. To see what the `top.sls` configuration looks like from that minion’s perspective, use `state.show_top`:

```
$ sudo salt new-client-dev state.show_top
new-client-dev:
  ------------
  development:
  - global
  - webserver
  - database
```

The output shows only the states that are active and selected for the target minion. In other words, they are the states that would run if you invoked `state.highstate` on that minion.

Note that all the displayed states come from the development environment. Because the minion is pinned, that will always be the case.
Documentation roadmap

Salt’s documentation (docs.saltstack.com) will likely earn your admiration, but perhaps only after a period of frustration. The main sticking point is that topics are nested several layers deep, but the headings at the top two layers do not necessarily hint at what you’ll find at layer three. The Architecture section, for example, contains no information about Salt’s architecture (it’s really about multiserver deployments).

Some of the most useful reference material lies within sections that are organized as scenarios or tutorials. Front-to-back reading can sometimes evoke a dying sysadmin’s fever dream: themes cyclically loom and recede without fully resolving. Once in a while, you’ll experience a moment of lucidity in which to appreciate the severity of your condition.

Some pointers:

- The top-level Using Salt section is an overview by concept, and most of Configuration Management is labeled as a tutorial. Because of their formats, these sections look like supplemental materials. But that’s not true; they are pretty much the primary documentation for the material they cover. Don’t skip.

- The best reference information is beneath State System Reference, under Configuration Management. A lot of the stuff in here is not important for a first reading, but Highstate data structure definitions, Requisites and other global state arguments, and The top file are particularly worth reading. (The top file is also the authoritative documentation for environments.)

- The docs you’ll use most frequently—the ones covering state and execution functions—are concealed under Salt Module Reference and camouflaged among 19 other module types that are of interest mostly to module developers. Bookmark the sections for Full list of builtin state modules and Full list of builtin execution modules.
23.7 Ansible and Salt Compared

We like both Ansible and Salt. Each of them has some friction points, however, and we recommend them for different environments. The sections below comment on a few of the factors you might consider when choosing between them.
Deployment flexibility and scalability

Salt covers a broader range of deployment environments than does Ansible. It’s simple enough that you can reasonably use it to manage a single server, but it also scales effortlessly and essentially without limit. If you want to learn one system that covers the broadest possible range of use cases, Salt is a good choice.

In part, that’s because Salt’s architecture makes relatively few demands of the master server. Minions receive their instructions and don’t report back until they’re done, with all status information being reported at once. Minions call the server to obtain configuration data, but aside from serving pillar data, the server itself performs relatively little computation.

Once your site outgrows a single Salt master, you can convert your infrastructure to a tiered or replicated server scheme. We don’t cover those options in this book, but they’re easy to set up and work well.

Large deployments are a comparative weak spot for Ansible. It does include some features to help you implement multitier server systems, but the transition to this model is not as transparent as it is in Salt.

Ansible is an order of magnitude slower than Salt, and because of its architecture, it must handle clients in batches. However, most servers can handle far more than the default 5 simultaneous clients. You can also change Ansible’s execution strategy so that clients aren’t kept in strict lockstep with each other. Even a tuned Ansible system won’t approach the speed of Salt, but it’s better than one might naively anticipate.
Built-in modules and extensibility

Bake-offs of configuration management software sometimes attempt to compare the number of operation types that various systems support out of the box. However, these comparisons are hard to get right because of underlying structural differences. Functions that are spread across several modules in Ansible might be addressed by one in Salt, for example. An atomic operation in one system may correspond to several operations in another.

At present, Salt and Ansible are roughly comparable in this respect. In addition to extensive standard libraries, both systems have a structure in place for absorbing community-written modules into the core or an easily accessible add-on pack.

In any event, total module count doesn’t matter nearly as much as coverage of the systems and software your site is actually using. All CM systems cover basic operations pretty well, but as you move into the long tail, offerings vary dramatically.

It’s likely that you’ll eventually want to tackle some tasks for which your CM system doesn’t have an off-the-shelf solution. Fortunately, Salt and Ansible are both easy to extend with your own Python code. Embrace this extensibility early on and make it part of your repertoire.
Security

As outlined in Ansible access options, Ansible can be made almost arbitrarily secure. The only limit to security is your own willingness to retype passwords and deal with security red tape.

Ansible’s vault system lets you keep configuration data in an encrypted format. That’s actually a pretty big deal, because it means that neither the Ansible server nor the configuration base needs to be particularly secure. (Salt’s modular architecture probably makes this an easy feature to add, but it doesn’t come in the box.)

By contrast, Salt can only be as secure as the root account on the master server. Although the master daemon itself is simple to set up, the server on which it runs should receive your site’s most aggressive securement. Ideally, the master should be a machine or virtual server dedicated to this task.

In practice, administrators hate intrusive security protocols as much as anyone else does. Most real-world Ansible installations have relatively lax security. Just as Ansible can be made arbitrarily secure, it can also be made arbitrarily insecure.

Even if you strive to keep Ansible fully secured, you may have trouble maintaining this approach once your site grows beyond the point at which configuration management can be handled by an administrator typing commands in a terminal window. Nothing that runs out of cron, for example, can depend on the presence of an administrator to enter passwords. Working around that constraint inevitably ends up lowering security to the level of the root account.

The bottom line on security is that Ansible gives you both more options and more opportunities to shoot yourself in the foot. It’s more securable, but that doesn’t necessarily mean that it’s more secure. Either system is fine for the average site. Keep your own needs and constraints firmly in mind when evaluating these systems.
Miscellaneous

Table 23.6 and Table 23.7 summarize some of the additional strengths and weakness of both Ansible and Salt.

Table 23.6: Ansible pros and cons

<table>
<thead>
<tr>
<th>Advantages</th>
<th>Disadvantages</th>
</tr>
</thead>
<tbody>
<tr>
<td>Requires only SSH and Python; no daemons</td>
<td>Very slow</td>
</tr>
<tr>
<td>Clear and concise documentation</td>
<td>Server-heavy; harder to scale</td>
</tr>
<tr>
<td>Built-in loops and conditionals, minimal Jinja</td>
<td>Lots of files with identical names</td>
</tr>
<tr>
<td>Works fine as a nonroot user</td>
<td>Idiosyncratic YAML syntax</td>
</tr>
<tr>
<td>Operations can use each other's output</td>
<td>Hand-managed client inventory</td>
</tr>
<tr>
<td>Clean and flexible use of config directories</td>
<td>Minimal grouping facilities</td>
</tr>
<tr>
<td>Secure, general encryption facility</td>
<td>Many different variable scopes</td>
</tr>
<tr>
<td>Roles can be instantiated repeatedly</td>
<td>No daemons means fewer options</td>
</tr>
<tr>
<td>Larger user base than Salt</td>
<td>No real support for environments</td>
</tr>
</tbody>
</table>

Table 23.7: Salt pros and cons

<table>
<thead>
<tr>
<th>Advantages</th>
<th>Disadvantages</th>
</tr>
</thead>
<tbody>
<tr>
<td>Fast</td>
<td>Relies heavily on Jinja</td>
</tr>
<tr>
<td>At heart, simpler than Ansible</td>
<td>Documentation oddly organized</td>
</tr>
<tr>
<td>Flexible, consistent bindings</td>
<td>Poor support for nonroot use</td>
</tr>
<tr>
<td>Integrated support for cloud servers</td>
<td>Formulas can't be instantiated</td>
</tr>
<tr>
<td>Concise configuration syntax</td>
<td>No built-in encryption solution</td>
</tr>
<tr>
<td>Multitier server deployments</td>
<td>No access to results of operations</td>
</tr>
<tr>
<td>Structured event monitoring</td>
<td>Minimal support for variable value defaults</td>
</tr>
<tr>
<td>Execution logs easily exported</td>
<td>Requires explicit dependency declarations</td>
</tr>
<tr>
<td>Fanatically modularized</td>
<td>Fanatically modularized</td>
</tr>
</tbody>
</table>
23.8 Best practices

If you’ve worked on a software project, you might find many of the issues addressed by configuration management systems to be familiar from the development world. Development environments encompass many of the same vagaries: multiple platforms, multiple products derived from the same code base, multiple types of builds and configurations, and deployment through successive steps of development, testing, and production.

These are complex issues, and development environments are only tools. Developers use a variety of additional controls—development guidelines, design reviews, coding standards, internal documentation, and clear architectural boundaries, among others—to limit the slide toward entropy.

Unfortunately, administrators often wander into configuration management territory without a proper suit of developer’s armor. At first glance, configuration management seems deceptively straightforward, like a slightly more general and sophisticated way to approach routine scripting tasks. Configuration management vendors work hard to reinforce this impression. Their web sites are siren songs of ease and grace; each one features a tutorial in which you deploy a web server by running ten lines of configuration code.

In reality, the edge of the abyss may be closer than it seems, particularly when multiple administrators contribute to the same configuration base over time. Real-world specifications for even a single-purpose server run to hundreds of lines of code segmented into multiple different functional roles. Without coordination, it’s easy to turn the CM system into a muddle of conflicting or parallel code.

Best practices vary by configuration management system and by environment, but a few rules apply to most situations:

- Keep the configuration base under version control. This isn’t a best practice so much as a basic requirement for CM sanity. Not only does Git provide change tracking and history, but it has already solved many of the mechanical problems involved in coordinating projects across administrative boundaries.

- Configuration bases are inherently hierarchical, at least in a logical sense. Some standards apply site-wide, some apply to every server in a particular department or region, and some are specific to particular hosts. In addition, you’ll most likely need the ability to make exceptions in certain cases. Depending on your site’s operations, you might also need to maintain multiple independent hierarchies.

Plan for all of this structure in advance, and consider how you might manage scenarios in which different groups control different parts of the configuration base. At the very least, conventions for classifying hosts (e.g., EC2 instances, Internet-
facing hosts, database servers) should be coordinated site-wide and adhered to consistently.

- CM systems allow different parts of the configuration base to be kept in different directories or repositories. However, this structure provides little actual benefit, and it complicates day-to-day configuration work. We recommend one big, integrated configuration base. Manage hierarchy and coordination at the Git layer.

- Sensitive data (keys, passwords) should not be put under version control unless encrypted, even in private code repositories. Git in particular is not designed to maintain security. Your CM system may have some encryption features built in, but if not, roll your own.

- Because it effectively has root access to many other hosts, a configuration server is one of the most concentrated sources of security risk in your organization. It’s reasonable to dedicate a server to this role, and it should receive your most stringent security hardening.

- Configurations should run without reporting spurious changes. Scripts and shell commands are usually the biggest sticking points. Check your CM system’s documentation for advice on this topic, as it’s one of the most frequent issues that users encounter.

- Don’t test on production servers. But do test! It’s easy to spin up a test system in the cloud or in Vagrant. Chef even provides an elaborate testing and development system in the form of Kitchen. Make sure your test system matches your real systems by using the same machine image and network configuration.

- Read the code for add-on bundles that you obtain from public repositories. It’s not that these sources are particularly suspicious; it’s just that systems and conventions differ widely. In many cases, you’ll find that a few local tweaks are needed. If you can bypass the CM system’s package manager and clone bundles directly from a Git repo, then you can easily upgrade to later releases without losing your customizations.

- Subdivide configurations ruthlessly. Every file should have a clear and single purpose. (Ansible users might want to select a text editor that deals well with 50 different files all named `main.yml`.)

- Configuration-managed servers should be 100% managed. That is, there should be no penumbra of administrative work that was performed by hand and that no one knows how to replicate. This issue appears primarily when moving existing servers onto configuration management. When converting an existing “snowflake” server to configuration management, you may find it useful to clone the original system for use as a basis for comparison. It can take multiple cycles of configuration management and testing to home in on all the system’s particulars.
- Do not allow yourself or your team to “temporarily disable” the CM system on a node or to use a heavy-handed method of overriding the CM system (for example, by setting the immutable attribute on a configuration file that is typically under CM control). These changes are inevitably forgotten, and confusion or outages ensue.

- It’s not hard to open a gateway from existing administrative databases into a configuration management system, and there’s a lot of value in doing so. CM systems are designed for this kind of interfacing. For example, you might identify system administrators and their zones of activity in your site-wide LDAP database, and make this information available within the configuration management environment through gateway scripts. Ideally, every piece of information should have a single authoritative source.

- CM systems are excellent for managing the state of a machine. They are not intended for stateful, coordinated activities such as software deployment operations, although the documentation and even some examples might lead you to believe that they are. In our experience, a dedicated continuous deployment system is more suitable.

- In elastic cloud environments where computational capacity is added in response to real-time demand, the time that it takes for a new node to bootstrap through configuration management can be agonizingly slow. Optimize by including packages and long-running configuration items within the baseline machine image rather than downloading and installing them at boot time.

  If you use configuration management to set configuration parameters for an application, make sure that that step comes early in the bootstrapping process so that the application comes on-line more quickly. We try to limit the CM run time to less than 60 seconds for dynamically scaled nodes.

- As an administrator working with a CM system, you will allot much of your time to writing CM code, testing changes against a representative set of systems, committing the updates to a repository, and applying changes to your site in a staged fashion. To be most effective, you should perfect this process by investing time up front to learn best practices and tricks for your system of choice.
23.9 **RECOMMENDED READING**


**Geerling, Jeff.** *Ansible for DevOps: Server and configuration management for humans.* St. Louis, MO: Midwestern Mac, LLC, 2015. This book is focused mostly on basic Ansible wrangling, but it does include some helpful material about combining Ansible with specific systems such as Vagrant, Docker, and Jenkins.

**Hochstein, Lorin.** *Ansible: Up and Running (2nd Edition).* Sebastopol, CA: O’Reilly Media, 2017. Like *Ansible for DevOps*, this book covers both the Ansible basics and interactions with common environments such as Vagrant and EC2. Some highlights are the inclusion of a larger-scale example configuration, a chapter on writing your own Ansible modules, and tips on debugging.

**Morris, Kief.** *Infrastructure as Code: Managing Servers in the Cloud.* Sebastopol, CA: O’Reilly Media, 2016. This book includes few specifics about configuration management per se, but it’s helpful for understanding how configuration management integrates into the larger scheme of DevOps and structured administration.

**Sebenik, Craig, and Thomas Hatch.** *Salt Essentials.* Sebastopol, CA: O’Reilly Media, 2015. This is a short and rather skeletal book that sticks pretty closely to the basics of Salt. It’s not the style of book we’d ordinarily recommend, but given the unevenness of the official documentation, it’s a potentially useful reference for those who seek a “second opinion.”


Server virtualization makes it possible to run multiple operating system instances concurrently on the same physical hardware. Virtualization software parcels out CPU, memory, and I/O resources, dynamically allocating their use among several “guest” operating systems and resolving resource conflicts. From the user’s point of view, a virtual server walks and talks like a full-fledged physical server.

See Chapter 25 for more information about containers.

This decoupling of hardware from the operating system affords numerous luxuries. Virtualized servers are more flexible than their “bare metal” kin. They’re portable and can be programmatically managed. The underlying hardware is used more efficiently because it can service multiple guests simultaneously. And if that isn’t enough, virtualization technology underpins both cloud computing and containers.

Implementations of virtualization have changed over the years, but the core concepts are not new to the industry. Big Blue used virtual machines on early mainframes while researching time-sharing concepts in the 1960s. The same techniques were used throughout the mainframe heyday
of the 1970s until the client/server boom of the 1980s, when the difficulty of virtualizing the Intel x86 architecture led to a short period of relative dormancy.

The ever-growing size of server farms rekindled interest in virtualization for modern systems. VMware and other providers conquered the challenges of x86 and made it easy to automatically provision operating systems. These facilities eventually led to the rise of on-demand, Internet-connected virtual servers: the infrastructure we now know as cloud computing. More recently, advances in OS-level virtualization have ushered in a new era of OS abstraction in the form of containers.

In this chapter, we begin by clarifying the terms and concepts you need in order to understand virtualization for UNIX and Linux. We then introduce the leading virtualization solutions used on our example operating systems.
24.1 Virtual Vernacular

The terminology used to describe virtualization is somewhat opaque, largely because of the way the technology evolved. Competing vendors worked independently without the benefit of standards, yielding a bewildering array of ambiguous phrases and acronyms. Conway’s Law comes to mind: “Organizations which design systems are constrained to produce designs which are copies of the communication structures of these organizations.”

To further confuse the issue, “virtualization” itself is an overloaded term that describes more than the scenario described above, in which guest operating systems run within the context of virtualized hardware. OS-level virtualization—more commonly referred to as containerization—is a related but distinct set of facilities that has become just as ubiquitous as server virtualization. For those without hands-on exposure to these technologies, it can often be difficult to grasp the differences. We contrast the two approaches later in this section.
Hypervisors

A hypervisor (also known as a virtual machine monitor) is a software layer that mediates between virtual machines (VMs) and the underlying hardware on which they run. Hypervisors are responsible for sharing system resources among the guest operating systems, which are isolated from one another and which access the hardware exclusively through the hypervisor.

Guest operating systems are independent, so they needn’t be the same. CentOS can run alongside FreeBSD and Windows, for example. VMware ESX, XenServer, and FreeBSD bhyve are examples of hypervisors. The Linux kernel-based virtual machine (KVM) converts the Linux kernel into a hypervisor.

Full virtualization

The first hypervisors fully emulated the underlying hardware, defining virtual replacements for all the basic computing resources: hard disks, network devices, interrupts, motherboard hardware, BIOSs, and so on. This mode, called full virtualization, runs guests without modification but incurs a performance penalty because the hypervisor must constantly translate between the system’s actual hardware and the virtual hardware exposed to guests.

Simulating an entire PC is a complex task. Most hypervisors that offer full virtualization separate the task of maintaining multiple environments (virtualization) from the task of simulating the hardware within each environment (emulation).

The most common emulation package used in these systems is an open source project called QEMU. You can find more information at qemu.org, but in most cases the emulator doesn’t require much attention from administrators.

Paravirtualization

The Xen hypervisor introduced “paravirtualization,” in which modified guest operating systems detect their virtualized state and actively cooperate with the hypervisor to access hardware. This approach improves performance by an order of magnitude or more. However, guest operating systems need substantial updates to run this way, and the exact modifications depend on the specific hypervisor in use.

Hardware-assisted virtualization

In 2004 and 2005, Intel and AMD introduced CPU features (Intel VT and AMD-V) that facilitate virtualization on the x86 platform. These extensions gave rise to “hardware-assisted virtualization,” also known as “accelerated virtualization.” In this scheme, the CPU and memory controller are virtualized by the hardware, albeit under the control of the hypervisor. Performance is very good, and guest operating systems need not know that they’re running on a virtualized CPU. These days, hardware-assisted virtualization is the assumed baseline.
Although the CPU is a primary point of contact between the hardware and guest operating systems, it is only one component of the system. The hypervisor still needs some way to present or emulate the rest of the system’s hardware. Either full virtualization or paravirtualization can be used for this task. In some cases, a mix of approaches is used; it depends on the virtualization-awareness of the guest.

**Paravirtualized drivers**

One great advantage of hardware-assisted virtualization is that it largely restricts the need for paravirtualization support to the level of device drivers. All operating systems allow add-on drivers, so setting up a guest with paravirtualized disk drives, display cards, and network interfaces is as simple as installing the appropriate drivers. The drivers know the secret handshake that lets them connect with the hypervisor’s paravirtualization support, and the guest OS remains none the wiser.

A few pesky aspects of the PC architecture, such as the interrupt controller and the BIOS resources, fall into the domain of neither the CPU nor the device drivers. In the past, the predominant approach has been to implement these remaining components through full virtualization. For example, Xen’s HVM (Hardware Virtual Machine) mode combines support for CPU-level virtualization extensions with a copy of the QEMU PC emulator. And PVHVM (ParaVirtualized HVM) mode adds to this scheme paravirtualized drivers on guest operating systems, greatly reducing the amount of full virtualization needed to keep the system running. However, the hypervisor still needs an active copy of QEMU for each virtual machine so that it can cover the odds and ends not addressed by the paravirtualized drivers.

**Modern virtualization**

The most recent versions of Xen and other hypervisors have more or less eliminated the need to emulate legacy hardware. Instead, they rely on CPU-level virtualization features, paravirtualized guest-OS drivers, and a few additional sections of paravirtualized code within guest kernels. Xen calls this mode PVH (ParaVirtualized Hardware), and it’s considered to be a close-to-ideal blend that yields optimal performance but imposes the lowest possible requirements on guest kernels.

You might encounter any of the varieties of virtualization described above in the wild or when reading documentation. However, it’s not worth memorizing any particular taxonomy or worrying too much about virtualization modes. The boundaries among these modes are porous, and the hypervisor generally works out the best options for a given guest. If you keep your software updated, you automatically benefit from the latest enhancements. The only reason to choose anything other than the default operating mode is to support older hardware or ancient hypervisors.

**Type 1 vs. type 2 hypervisors**

Many reference materials draw a somewhat dubious distinction between “type 1” and “type 2” hypervisors. A type 1 hypervisor runs directly on the hardware without a supporting OS, and for
that reason is sometimes called a bare-metal or native hypervisor. Type 2 hypervisors are user-space applications that run on top of another general-purpose OS. Exhibit A depicts these two models.

Exhibit A: Comparison of type 1 and type 2 hypervisors

<table>
<thead>
<tr>
<th>Type 1</th>
<th>Type 2</th>
</tr>
</thead>
<tbody>
<tr>
<td>Guest OS</td>
<td>Guest OS</td>
</tr>
<tr>
<td>Type 1 hypervisor</td>
<td>Type 2 hypervisor</td>
</tr>
<tr>
<td>Host OS</td>
<td></td>
</tr>
<tr>
<td>Memory</td>
<td>I/O</td>
</tr>
<tr>
<td>Hardware</td>
<td></td>
</tr>
</tbody>
</table>

VMware ESXi and XenServer are considered type 1, and FreeBSD’s bhyve is type 2. Likewise, workstation-oriented virtualization packages such as Oracle’s VirtualBox and VMware Workstation are also type 2.

It’s true that type 1 and type 2 systems are different, but the delineation is not always so binary. KVM, for example, is a Linux kernel module that gives virtual machines direct access to CPU virtualization features. Differentiating among types of hypervisor is more an academic exercise than a point of practice.
**Live migration**

Virtual machines can move between hypervisors running on different physical hardware in real time, in some cases without interruptions in service or loss of connectivity. This feature is called live migration. The magic lies in a memory dance between the source and target hosts. The hypervisor copies changes from the source to the destination, and as soon as the memory is identical between the two, the migration completes. Live migration is helpful for high-availability load balancing, disaster recovery, server maintenance, and general system flexibility.
Virtual machine images

Virtual servers are created from images, which are templates of configured operating systems that a hypervisor can load and execute. The image file format varies by hypervisor. Most hypervisor projects maintain a collection of images that you can download and use as a basis for your own customizations. You can also take a snapshot of a virtual machine to create an image, either as a backup of important data or to use as the basis for creating more virtual machines.

Because the virtual machine hardware presented by the hypervisor is standardized, images are portable among systems even if their actual hardware differs. Images are specific to a particular hypervisor, but conversion tools that port images among hypervisors are available.
Containerization

OS-level virtualization—or containerization—is a different approach to isolation that does not use a hypervisor. Instead, it relies on kernel features that isolate processes from the rest of the system. Each process “container” or “jail” has a private root filesystem and process namespace. The contained processes share the kernel and other services of the host OS, but they cannot access files or resources outside of their containers. Exhibit B illustrates this architecture.

See Chapter 25 for more information about containers.

Exhibit B: Containerization

Because it does not require virtualization of the hardware, the resource overhead of OS-level virtualization is low. Most implementations offer near-native performance. Linux’s LXC, Docker containers, and FreeBSD jails are implementations of containers.

This type of virtualization largely precludes the use of multiple operating systems because the host kernel is shared by all containers. However, FreeBSD’s Linux emulation layer permits Linux containers to run on FreeBSD hosts.

It’s easy to confuse containers with virtual machines. Both define portable, isolated execution environments, and both look and act like full operating systems with root filesystems and running processes. Yet their implementations are entirely different.

A true virtual machine has an OS kernel, an init process, drivers to interact with hardware, and the full trappings of a UNIX operating system. A container, on the other hand, is merely the facade of an operating system. It uses the strategies described above to give individual processes a suitable execution environment. Table 24.1 illustrates some of the practical differences.

Table 24.1: Comparing virtual machines with containers
<table>
<thead>
<tr>
<th>Virtual machine</th>
<th>Container</th>
</tr>
</thead>
<tbody>
<tr>
<td>A full-fledged OS that shares underlying hardware through a hypervisor</td>
<td>An isolated group of processes managed by a shared kernel</td>
</tr>
<tr>
<td>Requires a complete boot procedure to initialize; starts in 1-2 minutes</td>
<td>Processes run directly by the kernel; no boot required; starts in &lt; 1 second</td>
</tr>
<tr>
<td>Long-lived</td>
<td>Frequently replaced</td>
</tr>
<tr>
<td>Has one or more dedicated virtual disks attached through the hypervisor</td>
<td>Filesystem view is a layered construct defined by the container engine</td>
</tr>
<tr>
<td>Images measured in gigabytes</td>
<td>Images measured in megabytes</td>
</tr>
<tr>
<td>A few dozen or fewer per physical host</td>
<td>Many per virtual or physical host</td>
</tr>
<tr>
<td>Complete isolation among guests</td>
<td>OS kernel and services shared with host</td>
</tr>
<tr>
<td>Multiple independent operating systems running side by side</td>
<td>Must run the same kernel as the host (OS distribution may differ)</td>
</tr>
</tbody>
</table>

It’s common to use containers in combination with virtual machines. Virtual machines are the best way to subdivide physical servers into manageable chunks. You can then run applications in containers atop the VMs to achieve optimal system density (this procedure is sometimes called “bin packing”). The containers-on-VMs architecture is standard for containerized applications that need to run on public cloud instances.

We focus on true virtualization for the rest of this chapter. See [Chapter 25, Containers](#), for more details on containerization.
24.2 Virtualization with Linux

Xen and KVM are the leading open source virtualization projects for Linux. Xen, now a project of the Linux Foundation, powers some of the largest public clouds, including Amazon Web Services and IBM’s SoftLayer. KVM is the kernel-based virtual machine integrated into the mainline Linux kernel. Both Xen and KVM have demonstrated their stability through many production installations at large sites.
Xen

Initially developed by Ian Pratt as a research project at the University of Cambridge, the Linux-friendly Xen has grown to become a formidable virtualization platform that challenges even the commercial giants in terms of performance, security, and especially, cost.

As a paravirtual hypervisor, Xen claims a mere 0.1%–3.5% overhead, far less than fully virtualized solutions. Because Xen is open source, a variety of management tools are available with varying levels of feature support. The Xen source code is available from xenproject.org, but many Linux distributions include native support.

Xen is a bare-metal hypervisor that runs directly on the physical hardware. A running virtual machine is called a domain. There is always at least one domain, referred to as domain zero or dom0. Dom0 has full hardware access, manages the other domains, and runs all the hypervisor’s own device drivers. Unprivileged domains are referred to as domU.

Dom0 typically runs a Linux distribution. It looks just like any other Linux system but includes the daemons, tools, and libraries that complete the Xen architecture and enable communication among domU, dom0, and the hypervisor.

The hypervisor is responsible for CPU scheduling and memory management for the system as a whole. It controls all domains, including dom0. However, the hypervisor itself is in turn controlled and managed from dom0. What a tangled web we weave.

Table 24.2 lists the most interesting puzzle pieces of a Linux dom0.

Table 24.2: Xen components in dom0

<table>
<thead>
<tr>
<th>Path</th>
<th>Contents</th>
</tr>
</thead>
<tbody>
<tr>
<td>/etc/xen</td>
<td>Primary configuration directory</td>
</tr>
<tr>
<td>auto</td>
<td>Guest OS config files to autostart at boot time</td>
</tr>
<tr>
<td>scripts</td>
<td>Utility scripts that create network interfaces, etc.</td>
</tr>
<tr>
<td>/var/log/xen</td>
<td>Xen log files</td>
</tr>
<tr>
<td>/usr/sbin/xl</td>
<td>Xen guest domain management tool</td>
</tr>
</tbody>
</table>

Each Xen guest-domain configuration file in /etc/xen specifies the virtual resources available to a domU, including disk devices, CPU, memory, and network interfaces. Each domU has a separate configuration file. The format is flexible and gives administrators granular control over the constraints applied to each guest. If a symbolic link to a domU configuration file is added to the auto subdirectory, that guest OS is automatically started at boot time.
Xen guest installation

It takes several steps to get a guest server up and running under Xen. We recommend the use of a tool such as **virt-manager** (virt-manager.org) to simplify the process. **virt-manager** was originally a Red Hat project, but it has now been deproprietarized and is available for most Linux distributions. **virt-install**, its command-line OS provisioning tool, accepts installation media from a variety of sources, including SMB or NFS mounts, physical CDs or DVDs, and HTTP URLs.

Guest domains’ disks are normally stored in virtual block devices (VBDs) in dom0. The VBD can be connected to a dedicated resource such as a physical disk drive or logical volume. Or, it can be a loopback file, also known as a file-backed VBD, created with **dd**. Performance is better with a dedicated disk or volume, but files are more flexible and can be managed with normal Linux commands (such as **mv** and **cp**) in dom0. Backing files are sparse files that grow as needed.

Unless the system is experiencing performance bottlenecks, a file-backed VBD is usually the best choice. It’s a simple process to transfer a VBD onto a dedicated disk if you change your mind.

The installation of a guest domain might look like this:

```bash
$ sudo virt-install --name chef -f /vm/chef.img -l http://example.com/myos
  --ram 512 --nographics
```

This is a typical Xen guest domain with the name “chef,” a disk VBD location of `/vm/chef.img`, and installation media obtained through HTTP. The instance has 512MiB of RAM and uses no X Windows graphics support during installation. **virt-install** downloads the files needed to start the installation and then kicks off the installer process.

When the screen clears, install Linux through the standard text-based process, which includes network configuration and package selection. After the installation completes, the guest domain reboots and is ready for use. To disconnect from the guest console and return to dom0, just press <Control-]>.

It’s worth noting that although this example uses text-based installation, graphics-based installation through Virtual Network Computing (VNC) is also available.

**virt-install** saves the domain’s configuration in `/etc/xen/chef`. Here’s what it looks like:
You can see that the NIC defaults to bridged mode. In this case, the VBD is a "block tap" file that affords better performance than does a standard loopback file. The writable disk image file is presented to the guest as `/dev/xvda`.

The `xl` tool is convenient for day-to-day management of virtual machines. It lets you start and stop VMs, connect to their consoles, and investigate their current state. Below, we show the running guest domains, then connect to the console for the chef domU. IDs are assigned in increasing order as guest domains are created, and they are reset when the host reboots.

```
$ sudo xl list
   Name   ID Mem(MiB) VCPUs State  Time(s)
Domain-0  0 2502   2     r----- 397.2
cheif    19 512   1     -b---- 12.8
$ sudo xl console 19
```

To change the configuration of a guest domain (e.g., to attach another disk or to change the network to NAT mode instead of bridged), edit the guest’s configuration file in `/etc/xen` and reboot the guest.
KVM
KVM, the Kernel-based Virtual Machine, is a full virtualization platform that is the default for most Linux distributions. Like Xen’s HVM mode, KVM takes advantage of the Intel VT and AMD-V CPU extensions and relies (in a typical setup) on QEMU to implement a fully virtualized hardware system. Although the system is native to Linux, it has also been ported to FreeBSD as a loadable kernel module.

Since KVM defaults to full virtualization, many guest operating systems are supported, including Windows. Paravirtualized Ethernet, disk, and graphics card drivers are available for Linux, FreeBSD, and Windows. Their use is optional but recommended for performance.

Under KVM, the Linux kernel itself serves as the hypervisor. Memory management and scheduling are handled through the host’s kernel, and guest machines are normal Linux processes. Enormous benefits accompany this unique approach to virtualization. For example, the complexity introduced by multicore processors is handled by the kernel, and no hypervisor changes are required to support them. Linux commands such as `top`, `ps`, and `kill` show and control virtual machines, just as they would for other processes. The integration with Linux is seamless.
KVM guest installation

Although the technologies behind Xen and KVM are fundamentally different, the tools that install and manage guest operating systems are eerily similar. As under Xen, you can use `virt-install` to create new KVM guests. Use the `virsh` command to manage them.

The flags passed to `virt-install` vary slightly from those used for a Xen installation. To begin with, the `--hvm` flag says that the guest should be hardware virtualized, as opposed to paravirtualized. In addition, the `--connect` argument guarantees that the correct default hypervisor is chosen, since `virt-install` supports more than one hypervisor. Finally, the use of `--accelerate` is recommended, to take advantage of the acceleration capabilities in KVM. Ergo, a full command for installing an Ubuntu server guest from DVD-ROM looks something like this:

```
$ sudo virt-install --connect qemu:///system -n UbuntuYakkety
    --hvm -r 512 -f ~/ubuntu-Yakkety.img -s 12 -c /dev/dvd --os-type linux
    --accelerate --vnc
```

Would you like to enable graphics support? (yes or no)

Assuming that the Ubuntu installation DVD has been inserted, this command launches the installation and stores the guest in the file `~/ubuntu-Yakkety.img`, allowing it to grow to 12GB. Since we specified neither `--nographics` nor `--vnc`, `virt-install` asks whether to enable graphics.

The `virsh` utility spawns its own shell from which you can run commands. To open the shell, type `virsh --connect qemu:///system`. The following series of commands demonstrates some of the core functionality of `virsh`. Type `help` in the shell to see a complete list, or see the man page for the nitty-gritty details.

```
$ sudo virsh --connect qemu:///system

virsh list --all

<table>
<thead>
<tr>
<th>Id</th>
<th>Name</th>
<th>State</th>
</tr>
</thead>
<tbody>
<tr>
<td>3</td>
<td>UbuntuYakkety</td>
<td>running</td>
</tr>
<tr>
<td>7</td>
<td>CentOS</td>
<td>running</td>
</tr>
<tr>
<td>-</td>
<td>Windows2016Server</td>
<td>shut off</td>
</tr>
</tbody>
</table>

virsh start Windows2016Server
Domain WindowsServer started

virsh shutdown CentOS
Domain CentOS is being shutdown

virsh quit
```
24.3 FreeBSD bhyve

FreeBSD’s virtualization software is bhyve, a relatively new system first added in FreeBSD 10.0. It can run BSD, Linux, and even Windows guests. However, it runs on a limited set of hardware and is missing some of the core features found in other implementations.

With so many virtualization platforms that support FreeBSD on the market already, it’s unclear why the bhyve effort started when it did. Unless you are developing a custom platform that requires embedded FreeBSD virtualization, we recommend choosing another solution until this project matures.
24.4 VMware

VMware is the biggest player in the virtualization industry and was the first vendor to develop techniques to virtualize the fractious x86 platform. VMware is a commercial entity, but some of its products are free. They’re all worthy of consideration when you are choosing a site-wide virtualization technology.

The primary product of interest to UNIX and Linux administrators is ESXi, which is a bare-metal hypervisor for the Intel x86 architecture. The name stands for “Elastic Sky X, integrated”—you really can’t make this stuff up. ESXi is free (free like a box of puppies), but some useful functionality is limited to paid licensees.

In addition to ESXi, VMware offers some powerful, advanced products that facilitate centralized deployment and management of virtual machines. They also have the most mature live migration technology we’ve seen. In-depth coverage of the full VMware product suite is beyond the scope of this chapter, however.
24.5 VirtualBox

VirtualBox is a consumer-grade, cross-platform, type 2 hypervisor. It performs “probably good enough” virtualization of systems, typically for individuals. It’s popular among developers and end users because it is free, easy to install, easy to use, and often simplifies the creation and management of test environments.

Performance and hardware support are both weak points. VirtualBox is generally not suitable for “production” virtualization use, although its web site does describe it as a “professional” solution which is licensed for “enterprise” use. (This may in fact be the case with regard to Oracle operating systems, which are the only prebuilt VMs available.)

The history of VirtualBox is long and sordid. It originally began as a commercial product of Innotek GmbH but was released as open source before Innotek was acquired by Sun Microsystems in 2008. After Oracle swallowed Sun in 2010, the product was rebranded as Oracle VM VirtualBox. VirtualBox lives on today (available under the GPLv2 open source license) and remains under active development at Oracle.

VirtualBox runs on Linux, FreeBSD, Windows, macOS, and Solaris. Oracle does not publish or support the FreeBSD version of the host, but it’s available as a community port. Supported guest OSs include Windows, Linux, and FreeBSD.

By default, you wrangle virtual machines through VirtualBox’s GUI. If you’re interested in running VMs on a system that doesn’t run a GUI, explore VBoxHeadless, the morbid name for VirtualBox’s CLI tool. You can download VirtualBox and read more about it at virtualbox.org.
24.6 Packer

Packer (packer.io), from the esteemed open source company HashiCorp, is a tool for building virtual machine images from a specification file. It can build images for a variety of virtualization and cloud platforms. Integrating Packer into your workflow lets you be more or less virtualization-platform-agnostic. You can easily build your customized image for whatever platform you’re using on a given day.

To create an image, Packer launches an instance from a source image of your choosing. It then customizes the instance by running scripts or invoking other provisioning steps that you specify. Finally, it saves a copy of the virtual machine’s state as a new image.

This process is particularly helpful for supporting an “infrastructure as code” way of managing servers. Instead of manually applying changes to images, you modify a template that describes the image in abstract terms. You then check the specification into a repository as you would with traditional source code. This technique supplies you with outstanding transparency, repeatability, and reversibility. It also creates a clear audit trail.

Packer configurations are JSON files. Most administrators agree that JSON is a poor choice of format since it’s notoriously picky about quotes and commas and doesn’t allow comments. With luck, HashiCorp will soon convert Packer to their much improved custom configuration format, but until then you’re stuck editing JSON.

In a template, “builders” define how to create an image and “provisioners” configure and install software for the image. Builders exist for AWS, GCP, DigitalOcean, VMware, VirtualBox, and Vagrant, among others. Provisioners can be shell scripts, Chef cookbooks, Ansible roles, or other configuration management tools.

The following template, custom_ami.json, demonstrates AWS’s amazon-ebs builder and the shell provisioner.
Just as the CLI tool needs certain parameters to launch an instance, the `amazon-ebs` builder needs data such as API credentials, instance type, the source AMI on which to base the new image, and a VPC subnet where the instance should be located. Packer uses SSH to execute the provisioning step, so we make sure the instance has a public IP address.

In this case, the provisioner is a shell script called `customize_ami.sh`. Packer copies the script to the remote system with `scp` and runs it. There’s nothing special about such a script; it can do anything you’d normally do from a script. For example, it can add new users, download and configure software, or execute security hardening steps.

To create the AMI, invoke `packer build`:

```
$ packer build custom_ami.json
```

`packer build` notes each step of the creation process on the console. The `amazon-ebs` builder takes the following steps:

1. It automatically creates a key pair and a security group.
2. It starts the instance and waits for it to become accessible on the network.
3. It uses `scp` and `ssh` to perform the requested provisioning steps.

4. It creates an AMI by calling the EC2 CreateImage API.

5. It cleans up by terminating the instance.

If everything works correctly, Packer prints the AMI ID as soon as it is available for use. If a problem occurs during the build, Packer prints a magenta-colored error message and exits after cleaning up after itself.

The `-debug` argument to `packer build` pauses at each step to let you troubleshoot problems. You can also use the `null` builder to fix any errors without launching an instance each time you try to run a build.
24.7 Vagrant

Also developed by HashiCorp, Vagrant is a wrapper that sits on top of virtualization platforms such as VMware, VirtualBox, and Docker. However, it is not itself a virtualization platform.

Vagrant simplifies virtual environment provisioning and configuration. Its mission is to quickly and easily create disposable, preconfigured development environments that closely mirror production environments. This glue function lets developers write and test code with minimal involvement from sysadmins or an operations team.

It’s possible (but not required) to use Vagrant in combination with Packer. For example, you might standardize the base image that you use for your production platforms through Packer, then distribute a Vagrant build of that image to developers. The developers can then spin up an instance of the image on their laptop or cloud provider of choice, with any necessary customizations. This method balances the need for centralized management of production images with developers’ need for access to a similar environment that they can directly control.
24.8 RECOMMENDED READING

The web site virtualization.info is an excellent source of current news, trends, and gossip in the virtualization and cloud computing sectors.


SENTHIL, NATHAN. VirtualBox at Warp Speed: Virtualization with VirtualBox. Seattle, WA: Amazon Digital Services, 2015.

25 Containers

Few technologies have generated as much excitement and hype in recent years as the humble container, whose explosion in popularity coincided with the release of the open source Docker project in 2013. Containers are of particular interest to system administrators because they standardize software packaging, an ambition that has long been tantalizingly out of reach.

To illustrate the utility of containers, consider a typical web application developed in any modern language or framework. At a minimum, the following ingredients are needed to install and run the app:

- The code for the application and its correct configuration
- Libraries and other dependencies, potentially numbered in the dozens, each pinned to a specific version that is known to be compatible
- An interpreter (e.g., Python or Ruby) or run time (JRE) to execute the code, also version pinned
- Localizations such as user accounts, environment settings, and services provided by the operating system

A typical site runs dozens or hundreds of such applications. Maintaining uniformity in each of these areas across multiple application deployments is a constant challenge, even with the assistance of the tools discussed in Chapter 23, Configuration Management, and Chapter 26,
Continuous Integration and Delivery. Incompatible dependencies required by separate applications lead to systems that are underutilized because they cannot be shared. In addition, at sites where software developers and system administrators are functionally separated, careful coordination is needed because it’s not always straightforward to identify who’s responsible for what parts of the operating environment.

A container image simplifies matters by packaging an application and its prerequisites into a standard, portable file. Any host with a compatible container run-time engine can create a container by using the image as a template. Tens or hundreds of containers can run simultaneously without conflicts. With images typically being a few hundred megabytes in size or less, it’s practical to copy them among systems. This easy application portability is perhaps the primary reason for the popularity of containers.

This chapter focuses on Docker. The eponymous business behind Docker has played a central role in bringing containers into mainstream use, and the Docker ecosystem is the one you’re most likely to encounter as a system administrator. Docker, Inc., offers several products related to containers, but we limit our discussion to the main container engine and the Swarm cluster manager.

Several viable alternative container engines are available. rkt, from CoreOS, is the most complete. It has a cleaner process model than Docker and a more secure default configuration. rkt integrates well with the Kubernetes orchestration system. systemd-nspawn, from the systemd project, is another option for lightweight containers. It has fewer features than Docker or rkt, but in some cases that can be a good thing. rkt cooperates with systemd-nspawn to configure container namespaces.
25.1 Background and Core Concepts

The container’s rapid rise to grace can be attributed more to timing than to the emergence of any single technology. Containers are a fusion of numerous existing kernel features, filesystem tricks, and networking hacks. A container engine is the management software that pulls it all together.

In essence, a container is an isolated group of processes that are restricted to a private root filesystem and process namespace. The contained processes share the kernel and other services of the host OS, but by default they cannot access files or system resources outside their container. Applications that run within a container are not aware of their containerized state and do not require modification.

After you read the following sections, it should be clear that containers contain no magic. In fact, they rely on some features of UNIX and Linux that have been around for many years. See Chapter 24, Virtualization, for a description of how containers differ from virtual machines.
Kernel support

The container engine uses several kernel features that are essential for isolating processes. In particular:

- **Namespaces** isolate containerized processes from the perspective of several operating system facilities, including filesystem mounts, process management, and networking. The mount namespace, for example, shows processes a customized view of the filesystem hierarchy. Containers can run with varying levels of integration with the host operating system, depending on how these namespaces have been configured.

- **Control groups** (contextually abbreviated to cgroups) limit the use of system resources and prioritize certain processes over others. Cgroups prevent runaway containers from consuming all available CPU and memory.

- **Capabilities** allow processes to execute certain sensitive kernel operations and system calls. For example, a process might have a capability that permits it to change the ownership of a file or to set the system time.

- **Secure computing mode** (usually shortened to seccomp) restricts access to system calls. It allows more fine-grained control than do capabilities.

Development of these features was driven in part by the Linux Containers project, LXC, which began at Google in 2006. LXC was the basis of Borg, Google’s internal virtualization platform. LXC supplies the raw functions and tools needed to create and run Linux containers, but with more than 30 command-line tools and configuration files, it’s quite complicated. The first few releases of Docker were essentially user-friendly wrappers that made LXC easier to use.

Docker now relies on an improved, standards-based container run time dubbed **containerd**. It too relies on Linux namespaces, cgroups, and capabilities to isolate containers. Learn more at containerd.io.
Images

A container image is akin to a template for a container. Images rely on union filesystem mounts for performance and portability. Unions overlay multiple filesystems to create a single, consistent hierarchy. The LWN.net article “A brief history of union mounts” describes the relevant background. The related articles are interesting reading, too. See lwn.net/Articles/396020.

Container images are union filesystems that are organized to resemble the root filesystem of a typical Linux distribution. The directory layout and the locations of binaries, libraries, and supporting files all conform to standard Linux filesystem hierarchy specifications. Specialized Linux distributions have been developed for use as the basis of container images.

To create a container, Docker points to the read-only union filesystem of an image and adds a read/write layer that the container can update. When containerized processes modify the filesystem, their changes are transparently saved within the read/write layer. The base remains unmodified. This is known as a copy-on-write strategy.

Many containers can share the same immutable base layers, thus improving storage efficiency and reducing startup times. Exhibit A depicts the scheme.

Exhibit A: Docker images and the union filesystem
Networking

The default way to connect containers to the network is to use a network namespace and a bridge within the host. In this configuration, containers have private IP addresses that aren’t reachable from outside the host. The host acts as a poor man’s IP router and proxies traffic between the outside world and the containers. This architecture gives administrators control over which container ports are exposed to the outside world.

It’s also possible to forgo the private container addressing scheme and expose entire containers directly to the network. This is called host mode networking, and it means that the container has unfettered access to the host’s network stack. This might be desirable in some situations, but it also presents a security risk because the container is not fully isolated.

See Docker networks for more details.
25.2 **Docker: the open source container engine**

Docker, Inc.’s primary product is a client/server application that builds and manages containers. The Docker container engine, written in Go, is highly modular. Separate, individual projects manage pluggable storage, networking, and other features.

Docker, Inc., is not without controversy. Its tools tend to evolve rapidly, and new versions have sometimes been incompatible with existing deployments. Some sites worry that relying on Docker’s ecosystem will result in vendor lock-in. And as with any new technology, containers introduce complexity and require some study to understand.

To counter these sources of resistance, Docker, Inc., became one of the founding members of the Open Container Initiative, a consortium whose mission is to guide the growth of container technology in a healthily competitive direction that fosters standards and collaboration. You can learn more at opencontainers.org. In 2017, Docker founded the Moby project and contributed the primary Docker Git repository to it to facilitate easier community development of the Docker execution engine. Refer to mobyproject.org for details.

Our discussion of Docker is based on version 1.13.1. Docker maintains an exceptionally rapid pace of development, and the current features are a moving target. We focus on the nuts and bolts here, but be sure to supplement our tutorial with the reference material at docs.docker.com. You might also dip your toes into the Moby sandbox at play-with-moby.com and the Docker lab environment at labs.play-with-docker.com.
Basic architecture

docker is an executable command that handles all management tasks for the Docker system. dockerd is the persistent daemon process that implements container and image operations. docker can run on the same system as dockerd and can communicate with it through UNIX domain sockets, or it can contact dockerd from a remote host over TCP. The architecture is depicted in Exhibit B.

Exhibit B: Docker architecture
**dockerd** owns all the scaffolding needed to run containers. It creates the virtual network plumbing and maintains the data directory in which containers and images are stored (/var/lib/docker by default). It’s responsible for creating containers by invoking the appropriate system calls, setting up union filesystems, and executing processes. In short, it is the container management software.

Administrators interface with **dockerd** from the command line by running subcommands of the **docker** client. You can create a container with **docker run**, for example, or view information about the server with **docker info**. **Table 25.1** summarizes some frequently used subcommands.

**Table 25.1: Frequently used docker subcommands**
An image is the *template* for a container. It includes the files that processes running within the container instance depend on, such as libraries, operating system binaries, and applications. Linux distributions can function as convenient base images because they define a complete operating environment. However, an image is not necessarily based on a Linux distribution. The “scratch” image is an explicitly empty image intended as a basis for creation of other, more practical images.

A container relies on the image template as a basis for execution. When `dockerd` runs a container, it creates a writable filesystem layer that is separate from the source image. The container can read any of the files and other metadata stored within the image, but any writes are confined to the container’s own read/write layer.

An image registry is a centralized collection of images. `dockerd` communicates with registries when you `docker pull` an image that isn’t already present or when you `docker push` one of your own images. The default registry is Docker Hub, which stockpiles images for many popular applications. Most standard Linux distributions also publish Docker images.

You can run your own registry, or you can add your custom images to private registries that are hosted on Docker Hub. Any system with Docker can pull images from a registry as long as the registry server is accessible over the network.
Installation

Docker runs on Linux, macOS, Windows, and FreeBSD, but Linux is the flagship platform. FreeBSD support is considered experimental. Visit docker.com to choose the installation method that best suits your environment.

Users in the docker group can control the Docker daemon through its socket, which effectively gives those users root privileges. This is a significant security risk, so we suggest that you use `sudo` to control access to `docker` rather than adding users to the docker group. In the examples below, we run `docker` commands as the root user.

The installation process may not immediately start the daemon. If it isn’t running, start it through the system’s normal `init` system. On CentOS, for example, run `sudo systemctl start docker`. 
**Client setup**

If you’re connecting to a local `dockerd` and you’re in the docker group or have `sudo` privileges, no client configuration is necessary. The `docker` client connects to `dockerd` through a local socket by default. You can modify the default client behavior by setting environment variables.

To connect to a remote `dockerd`, set the `DOCKER_HOST` environment variable. The usual HTTP port for the daemon is 2375, and the TLS version is 2376.

For example:

```bash
$ export DOCKER_HOST=tcp://10.0.0.10:2375
```

Always use TLS to communicate with remote daemons. If you use plain HTTP, you may as well hand out root privileges freely to anyone on your network. You can find additional details on Docker TLS configuration in [Use TLS](#).

We also suggest enabling the content trust:

```bash
$ export DOCKER_CONTENT_TRUST=1
```

This feature validates the integrity and publisher of Docker images. Enabling the content trust prevents the client from pulling images that are not trusted.

If you run `docker` through `sudo`, you can prevent `sudo` from purging your environment variables with the `-E` flag. You can also whitelist specific environment variables by setting the value of the `env_keep` variable in `/etc/sudoers`. For example,

```bash
Defaults env_keep += "DOCKER_CONTENT_TRUST"
```
The container experience

To create a container, you need an image to use as a template. The image has all the filesystem bits needed to run programs. A new installation of Docker has no images. You can review the available images by browsing hub.docker.com. To download images from the Docker Hub, use `docker pull`.

```
# docker pull debian
Using default tag: latest
latest: Pulling from library/debian
f50f9524513f: Download complete
d8bd0657b25f: Download complete
Digest: sha256:e7d38b351754b1a1c71e41bffe9c8ae6d6...
Status: Downloaded newer image for debian:latest
```

The hex strings are the layers of the union filesystem. If the same layer is used by more than one image, Docker needs only a single copy. We didn’t request a specific tag, or version, of the Debian image, so Docker downloaded the “latest” tag by default.

Examine the locally available images with `docker images`:

```
# docker images
REPOSITORY  TAG       IMAGE ID          CREATED          SIZE
ubuntu      latest    07c86167c6c4  2 weeks ago    187.9 MB
ubuntu      wily      b5e09e0cd052  5 days ago     136.1 MB
ubuntu      trusty   97434d46f197  5 days ago     187.9 MB
ubuntu      15.04     d1b55fd07600  8 weeks ago    131.3 MB
centos      7         d0e7f81ca65c  2 weeks ago    196.6 MB
centos      latest    d0e7f81ca65c  2 weeks ago    196.6 MB
debian      jessie    f50f9524513f  3 weeks ago    125.1 MB
debian      latest    f50f9524513f  3 weeks ago    125.1 MB
```

This machine has the images for several Linux distributions, including the just-downloaded Debian image. The same image can be tagged more than once. Notice that debian:jessie and debian:latest share an image ID; they are two different names for the same image.

Armed with an image, it’s remarkably simple to run a basic container:

```
# docker run debian /bin/echo "Hello World"
Hello World
```

What just happened? Docker created a container from the Debian base image and ran the command `/bin/echo "Hello World"` inside it. (This is GNU `echo`, not to be confused with the `echo` command built into most shells. They do exactly the same thing.)

The container stops running when the command exits: in this case, immediately after `echo` completes. If the “debian” image didn’t already exist locally, the daemon would attempt to
automatically download it before running the command. We didn’t specify a tag, so the “latest” image was used by default.

We start an interactive shell with the -i and -t flags to **docker run.** The command below starts a **bash** shell within the container and connects the “outer” shell’s I/O channels to it. We also assign the container a hostname, which is helpful for identifying it in logs. (Otherwise, we’d see the container’s random ID in log messages.)

```bash
ben@host$ sudo docker run --hostname debian -it debian /bin/bash
```

```bash
root@debian:/# ls
bin  dev  home  lib64  mnt  proc  run  srv  tmp  var
boot  etc  lib  media  opt  root  sbin  sys  usr
root@debian:/# ps aux
USER  PID   %CPU  %MEM  VSZ   RSS   TTY  STAT START  TIME COMMAND
root  1    0.5  0.4    29236  1884  ?    Ss   19:02  0:00 /bin/bash
root  7    0.0  0.2    17492  1144  ?    R+   19:02  0:00 ps aux
```

```bash
root@debian:/# uname -r
3.10.0-327.10.1.el7.x86_64
root@debian:/# exit
```

```bash
exit
ben@host$ uname -r
3.10.0-327.10.1.el7.x86_64
```

The experience is oddly similar to accessing a virtual machine. There is a complete root filesystem, but the process tree appears nearly empty. **/bin/bash** is PID 1 because it’s the command that Docker started in the container.

The result of **uname -r** is the same both inside and outside the container. That will always be the case; we show it as a reminder that the kernel is shared.

Processes in containers cannot see other processes running on the system because of PID namespacing. However, processes on the host can see the containerized processes. The PID of a process as seen from within a container differs from the PID that is visible from the host.

For real work, you need long-lived containers that run in the background and accept connections over the network. The following command runs in the background (-d) a container named “nginx” that’s generated from the official NGINX image. We tunnel port 80 from the host into the same port within the container:
We didn’t have the “nginx” image locally, so Docker had to pull it from the registry. Once the image was downloaded, Docker started the container and printed its ID, a unique 65-character hexadecimal string.

```
docker ps
```
shows a brief summary of running containers:

```
# docker ps

<table>
<thead>
<tr>
<th>IMAGE</th>
<th>COMMAND</th>
<th>STATUS</th>
<th>PORTS</th>
</tr>
</thead>
<tbody>
<tr>
<td>nginx</td>
<td>&quot;nginx -g 'daemon off&quot;</td>
<td>Up 2 minutes</td>
<td>0.0.0.0:80-&gt;80/tcp</td>
</tr>
</tbody>
</table>
```

We didn’t tell `docker` what to run in the container, so it used the default command that was specified when the image was created. The output shows this command to be `nginx -g 'daemon off'` which runs `nginx` as a foreground process rather than as a background daemon. The container has no `init` to manage processes, and if the `nginx` server were started as a daemon, the container would run but immediately exit when the `nginx` process forked and exited to enter the background.

Most server daemons offer a command-line flag that forces them to run in the foreground. If your software doesn’t run in the foreground or if you need to run several processes in a container, you can assign a process control system such as `supervisord` to act as a lightweight `init` for the container.

With NGINX running in the container and port 80 mapped from the host, we can make HTTP requests to the container with `curl`. NGINX serves a generic HTML landing page by default.

```
host$ curl localhost
<!DOCTYPE html>
<html>
<head>
<title>Welcome to nginx!</title>
...
```

We can use `docker logs` to view the STDOUT from the container, which in this case is the NGINX access log. The only traffic is our `curl` request:
We can also use `docker logs -f` to get a real-time stream of a container’s output, just like running `tail -f` on a growing log file.

`docker exec` creates a new process in an existing container. For example, to debug or troubleshoot, we could start an interactive shell in a container:

```
# docker exec -ti nginx bash
root@nginx:/# apt-get update & & apt-get -y install procps
root@nginx:/# ps ax
```

```
PID TTY STAT TIME COMMAND
1 ? Ss 0:00 nginx: master process nginx -g daemon off;
7 ? S 0:00 nginx: worker process
8 ? Ss 0:00 bash
21 ? R+ 0:00 ps ax
```

Container images are as slim as possible and are often missing common administrative utilities. In this sequence, we first updated the package index and then installed `ps`, which is part of the `procps` package.

The process list reveals the `nginx` master daemon, an `nginx` worker, and our `bash` shell. When we exit the shell created with `docker exec`, the container continues to run. If PID 1 exited while our shell was active, the container would terminate and our shell would also exit.

We can stop and start the container:

```
# docker stop nginx
nginx
# docker ps
IMAGE COMMAND STATUS PORTS
# docker start nginx
# docker ps
IMAGE COMMAND STATUS PORTS
nginx "nginx -g 'daemon off" Up 2 minutes 0.0.0.0:80->80/tcp
```

`docker start` starts the container with the same arguments that were passed when the container was created with `docker run`.

When containers exit, they remain on the system in a dormant state. You can list all containers, including those that are stopped, with `docker ps -a`. It’s not particularly harmful to keep unneeded old containers lying around, but it’s considered poor hygiene and might cause name collisions if you reuse container names.

When we finish with the container, we can stop and remove it:
# docker stop nginx && docker rm nginx

docker run --rm runs a container and removes it automatically when it exits, but this works only for containers that are not daemonized with -d.
Volumes

The filesystem layers for most containers consist of static application code, libraries, and other supporting or OS files. The read/write filesystem layer allows containers to make local modifications to these layers. However, heavy reliance on the overlay filesystem isn’t the best storage solution for data-intensive applications such as databases. For those kinds of apps, Docker has the notion of volumes.

A volume is an independent, writable directory within a container that’s maintained separately from the union filesystem. If the container is removed, the data in the volume persists and can be accessed from the host. Volumes can also be shared among multiple containers.

We add a volume to a container with `docker`’s `-v` argument:

```bash
# docker run -v /data --rm --hostname web --name web -d nginx
```

If `/data` already exists within the container, any files found there are copied to the volume. We can find the volume on the host by running `docker inspect`:

```bash
# docker inspect -f '{{json.Mounts}}' web
```

```

```

The `inspect` subcommand returns verbose output; we applied a filter so that only the mounted volumes would be printed. If the container terminates or needs to be removed, we can find the data volume at the Source directory on the host. The Name looks more like an ID, but it’s useful if we need to identify the volume later.

For a higher-level overview of volumes on the system, we run `docker volume ls`.

Docker also supports “bind mounts,” which mount volumes on the host and in containers simultaneously. For example, we can bind-mount `/mnt/data` from the host to `/data` in the container with the following command:

```bash
# docker run -v /mnt/data:/data --rm --name web -d nginx
```

When the container writes to `/data`, the changes are also visible in `/mnt/data` on the host.
For bind-mounted volumes, Docker does not copy existing files from the container’s mount directory to the volume. As with a traditional filesystem mount, the volume’s contents supersede the original contents of the container’s mount directory.

When running containers in the cloud, we suggest combining bind mounts with the block storage options offered by cloud providers. For example, AWS’s Elastic Block Storage volumes make great backing stores for Docker bind mounts. They have built-in snapshot facilities and can move among EC2 instances. They can also be copied between nodes, which makes it straightforward for other systems to retrieve a container’s data. You can leverage EBS’s native snapshotting facilities to create a simple backup system.
**Data volume containers**

One helpful pattern that has emerged from real-world experience is the data-only container. Its purpose is to hold a volume configuration on behalf of other containers so that those containers can be easily restarted and replaced.

Create a data container by using either a normal volume or a bind-mounted volume from the host. The data container never actually runs.

```bash
# docker create -v /mnt/data:/data --name nginx-data nginx
```

Now you can use the data container’s volume in the nginx container:

```bash
# docker run --volumes-from nginx-data -p 80:80 --name web -d nginx
```

The “web” container has read and write access to the `/data` volume of the data-only “nginx-data” container. “web” can be restarted, removed, or replaced, but so long as it is started with `--volumes-from`, the files in `/data` will remain persistent.

In truth, combining data persistence with containers is a bit of an impedance mismatch. Containers are meant to be created and removed at a moment’s notice in response to external events. The ideal is to have a fleet of identical servers that run `dockerd`, with containers being deployable to any of the servers. Once you add persistent data volumes, however, the container becomes coupled to a particular server. As much as we’d like to be living in the ideal world, many applications do need persistent data.
Docker networks

As discussed in Networking, there is more than one way to connect containers to the network. During installation, Docker creates three default networking options. List them with docker network ls:

```
# docker network ls
NETWORK ID  NAME           DRIVER
6514e71a8508 bridge       bridge
1a72c1e4b3c0 none          null
e0f4e608c92c host          host
```

See this page for more information about iptables.

In the default bridge mode, containers reside on a private namespaced network within the host. The bridge connects the host’s network to the container namespace. When you create a container and map a port from the host with docker run -p, Docker creates iptables rules that route traffic from the host’s public interface to the container’s interface on the bridge network.

With “host” networking, no separate network namespace is used. Instead, the container shares the network stack with the host, including all its interfaces. Ports exposed by the container are also exposed on the interfaces of the host. Some software behaves better when running with host networking, but this configuration can also lead to port conflicts and other problems.

“None” networking indicates that Docker shouldn’t take any steps whatsoever to configure networking. It is intended for advanced use cases that have custom networking requirements.

Pass the --net argument to docker run to select a container’s network.

Namespaces and the bridge network

A bridge is a Linux kernel feature that connects two network segments. During installation, Docker quietly creates a bridge called docker0 on the host. Docker chooses an IP address space for the far side of the bridge that it calculates as unlikely to collide with any networks reachable by the host. Each container is given a namespaced virtual network interface that has an IP address within the bridged network range.

The address selection algorithm is practical but not perfect. Your network may have routes that aren’t visible from the host. If a collision occurs, the host will no longer be able to access the remote network that has the overlapping address space, but it will be able to reach local containers. If you find yourself in this situation or if you need to customize the bridge’s address space for some other reason, use the --fixed-cidr argument to dockerd.

Network namespaces rely on virtual interfaces, strange constructs that are created in pairs, where one side is in the host’s namespace and the other is in the container’s. Data flows in one end of
the pair and out the other end, thus connecting the container to the host. In most cases a container has only one such pair. Exhibit C illustrates the concept.

**Exhibit C: A docker bridge network**

One half of each pair is visible from the host’s networking stack. For example, here are the visible interfaces on a CentOS host with just one container running:

```bash
centos$ ip addr show
2: enp0s3: <BROADCAST,MULTICAST,UP,LOWER_UP> mtu 1500 qdisc pfifo_fast
    state UP qlen 1000
    link/ether 08:00:27:c3:36:f0 brd ff:ff:ff:ff:ff:ff
    inet 10.0.2.15/24 brd 10.0.2.255 scope global dynamic enp0s3
        valid_lft 71368sec preferred_lft 71368sec
    inet6 fe80::a00:27ff:fe33:36f0/64 scope link
        valid_lft forever preferred_lft forever
3: docker0: <BROADCAST,MULTICAST,UP,LOWER_UP> mtu 1500 qdisc noqueue
    state UP
    link/ether 02:42:d4:30:59:24 brd ff:ff:ff:ff:ff:ff
    inet 172.17.42.1/16 scope global docker0
        valid_lft forever preferred_lft forever
    inet6 fe80::d4:ff:fe0b:bfeb/64 scope link
        valid_lft forever preferred_lft forever
53: veth584a021@if52: <BROADCAST,MULTICAST,UP,LOWER_UP> mtu 1500 qdisc noqueue master docker0 state UP
    link/ether d6:39:a7:bd:bf:eb brd ff:ff:ff:ff:ff:ff link-netnsid 0
    inet6 fe80::d439:a7ff:febd:bfeb/64 scope link
        valid_lft forever preferred_lft forever
```

The output shows enp0s3, the primary interface on the host, and docker0, the virtual Ethernet bridge, which uses the 172.17.42.0/16 range. The veth interface is the host side of the virtual interface pair that connects the container to the bridged network.

The container’s side of the bridged pair is not visible from the host without low-level inspection of the networking stack. This invisibility is just a side effect of the way network namespaces
work. However, we can find the interface by inspecting the container itself:

```
# docker inspect -f '{json.NetworkSettings.Networks.bridge}' nginx

  "bridge": {
    "Gateway": "172.17.42.1",
    "IPAddress": "172.17.42.13",
    "IPPrefixLen": 16,
    "MacAddress": "02:42:ac:11:00:03"
  }
```

The container’s IP address is 172.17.42.13, and the default gateway is the docker0 bridge interface. (This is the bridge network depicted in Exhibit C.)

In the default bridge configuration, all containers can communicate with one another because they are all on the same virtual network. However, you can create additional network namespaces to isolate containers from one another. That way, you can serve multiple, isolated environments from the same set of container instances.

**Network overlays**

Docker has lots of additional networking flexibility available to help with advanced use cases. For example, you can create user-defined private networks that automatically have container linking. With network overlay software, containers running on separate hosts can route traffic to each other through a private network address space. Virtual eXtensible LAN (VXLAN) technology, described in RFC7348, is one system that can be combined with containers to implement advanced networking capabilities. See the Docker networking documentation for more details.
Storage drivers

UNIX and Linux systems offer multiple ways to implement a union filesystem. Docker is technology-agnostic in this regard and filters all filesystem operations through a storage driver that you select.

The storage driver is configured as part of the docker daemon launch options. Your choice of storage engine has important consequences for performance and stability, especially in production environments that support many containers. Table 25.2 shows the current menu of drivers.

<table>
<thead>
<tr>
<th>Driver</th>
<th>Description and comments</th>
</tr>
</thead>
</table>
| aufs   | A reimplementation of the original UnionFS  
The original Docker storage engine  
Default for Debian and Ubuntu  
Now deprecated because it’s not part of the mainline Linux kernel |
| btrfs  | Uses the Btrfs copy-on-write filesystem  
Btrfs is stable and is included in the mainline Linux kernel  
Docker’s use is distribution-limited and somewhat experimental |
| devicemapper | Default for RHEL/CentOS 6  
Direct LVM mode strongly recommended but needs configuration  
Has a history of bugs  
Study Docker’s devicemapper documentation |
| overlay | Based on OverlayFS  
Considered the replacement for AuFS  
The default in CentOS 7 if the overlay kernel module is loaded |
| vfs    | Not a real union filesystem  
Slow but stable, suitable for some production environments  
Good as a proof of concept or as a testbed |
| zfs    | Uses the ZFS copy-on-write filesystem  
Default for FreeBSD  
Considered experimental on Linux |

The VFS driver effectively disables the use of a union filesystem. Docker creates a complete copy of an image for each container, resulting in higher disk usage and longer container start times. However, this implementation is simple and robust. If your use case involves long-lived containers, VFS is a reliable choice. We’ve never encountered a site that uses VFS in production, however.
Btrfs and ZFS are also not true union filesystems. However, they implement overlays efficiently and reliably because they natively support copy-on-write filesystem clones. Docker support for Btrfs and ZFS is currently limited to a few specific Linux distributions (and FreeBSD, for ZFS), but these are good options to keep an eye on for the future. The less filesystem magic specific to the container system, the better.

Storage driver selection is a nuanced topic. Unless you or somebody on your team has comprehensive knowledge of one of these filesystems, we recommend that you stick with the default for your distribution. The Docker storage driver documentation has further information.
**dockerd option editing**

You’ll inevitably need to modify some of `dockerd`’s settings. Tuning options include the storage engine, DNS options, and the base directory in which images and metadata are stored. Run `dockerd -h` to see a complete list of arguments.

You can examine a running daemon’s configuration with `docker info`:

```
centos# docker info
Containers: 6
   Running: 0
   Paused: 0
   Stopped: 6
Images: 9
Server Version: 1.10.3
Storage Driver: overlay
   Backing Filesystem: xfs
Logging Driver: json-file
Plugins:
   Volume: local
   Network: bridge null host
Kernel Version: 3.10.0-327.10.1.el7.x86_64
Operating System: CentOS Linux 7 (Core)
OS Type: linux
Architecture: x86_64
```

This is a good place to check that any customizations you made have taken effect.

Docker conforms to the operating system’s native `init` system for managing daemon processes, including settings for startup options. For example, on a distribution that uses `systemd`, the following command edits the Docker service unit to set a nondefault storage driver, a set of DNS servers, and a custom address space for the bridge network:

```
$ systemctl edit docker
[Service]
ExecStart=
ExecStart=/usr/bin/docker daemon -D --storage-driver overlay \
   --dns 8.8.8.8 --dns 8.8.4.4 --bip 172.18.0.0/19
```

The redundant `ExecStart=` is not a mistake. It’s a `systemd`-ism that clears the default setting, ensuring that the new definition is used exactly as shown. Once the edits are complete, we restart the daemon with `systemctl` and review the changes:
On systems running `upstart`, configure daemon options in `/etc/default/docker`. For older systems with SysV-style `init`, use `/etc/sysconfig/docker`.

By default, `dockerd` listens for connections from `docker` on the UNIX domain socket at `/var/run/docker.sock`. To set the daemon to listen on a TLS socket instead, use the daemon option `-H tcp://0.0.0:2376`. See `this page` for more details about how to set up TLS.
Image building

You can containerize your own applications by building images that include your application code. The build process begins with a base image. You add your application by committing any changes as new layers and saving the image to the local image database. You can then create containers from the image. You can also push your image to a registry to make it accessible to other systems running Docker.

Each layer of an image is identified by a cryptographic hash of its contents. The hash serves as a validation system that lets Docker confirm that no corruption or malicious intervention has modified the contents of the image.

Choosing a base image

Before creating a custom image, choose a suitable base. The rule of thumb for base images is that the smaller footprint, the better. The base should have what you need to run your software and nothing more.

Many of the official images are based on a distribution called Alpine Linux, which weighs in at a lean 5MB but may have library incompatibilities with some applications. The Ubuntu image is larger at 188MB, but still small in comparison to a typical server installation. You might be able to find a base image that has your application run-time components already configured. Default base images exist for the most common languages, run-times, and application platforms.

Thoroughly vet your base image before you make a final decision. Examine the base’s Dockerfile (see the next section) and any nonobvious dependencies to avoid surprises. Base images may have unexpected requirements or include vulnerable versions of software. In some circumstances, you may need to copy the Dockerfile of a base image and rebuild it to suit your needs.

When dockerd downloads an image, it downloads only the layers that it doesn’t already have. If all your applications use the same base, there is less data for the daemon to download and containers start faster when first run.

Building from a Dockerfile

A Dockerfile is a recipe for building an image. It contains a series of instructions and shell commands. The docker build command reads the Dockerfile, runs its instructions in sequence, and commits the result as an image. Software projects that have a Dockerfile usually keep it in the root directory of the Git repository to facilitate building new images that contain that software.

The first instruction in a Dockerfile specifies an image to use as the base. Each subsequent instruction commits a change to a new layer, which is used in turn as the base for the next
instruction. Each layer includes only the changes from the previous layer. The union filesystem merges the layers to create a container’s root filesystem.

Here is a Dockerfile that builds the official NGINX image for Debian. This is slightly simplified from github.com/nginxinc/docker-nginx.

FROM debian:jessie
MAINTAINER NGINX Docker Maintainers "docker-maint@nginx.com"
ENV NGINX_VERSION 1.10.3-1~jessie
RUN apt-get update \
    && apt-get install -y ca-certificates nginx=${NGINX_VERSION} \
    && rm -rf /var/lib/apt/lists/*
# forward request and error logs to docker log collector
RUN ln -sf /dev/stdout /var/log/nginx/access.log \
    && ln -sf /dev/stderr /var/log/nginx/error.log
EXPOSE 80 443
CMD ["nginx", "-g", "daemon off;"]

NGINX uses the debian:jessie image as a base. After declaring a maintainer, the file sets an environment variable (NGINX_VERSION) that’s then available to every subsequent instruction in the Dockerfile and also to any process that runs inside the container once the image has been built and instantiated. The first RUN instruction does the heavy lifting by installing NGINX from a package repository.

By default, NGINX sends log data to /var/log/nginx/access.log, but the convention for containers is to log messages to STDOUT. In the final RUN command, the maintainers use a symbolic link to redirect the access log to the STDOUT device file. Similarly, errors are redirected to the container’s STDERR.

The EXPOSE command tells dockerd which ports the container listens on. The exposed ports can be overridden at container run time with the -p option to docker run.

The final instruction in the NGINX Dockerfile is the command that dockerd should execute when it starts the container. In this case, the container runs the nginx binary as a foreground process.

See Table 25.3 for a rundown of common Dockerfile instructions. The reference manual at docs.docker.com is the authoritative documentation.

Table 25.3: Abbreviated list of Dockerfile instructions
Composing a derived **Dockerfile**

We can use a very simple **Dockerfile** to build a derived NGINX image that adds a custom **index.html**, replacing the default from the official image:

```bash
$ cat index.html
<!DOCTYPE html>
<title>ULSAH index.html file</title>
<p>A simple Docker image, brought to you by ULSAH.</p>
$ cat Dockerfile
FROM nginx
# Add a new index.html to the document root
ADD index.html /usr/share/nginx/html/
```

Other than having a custom **index.html**, our new image will be identical to the base image. Here's how we build the customized image:

```
# docker build -t nginx:ulsah .
Step 1 : FROM nginx
    ---> fd19524415dc
Step 2 : ADD index.html /usr/share/nginx/html/  
    ---> c0c25eaf7415
Removing intermediate container 04cc3278fde4
Successfully built c0c25eaf7415
```
We use `docker build` with `-t nginx:ulsah` to create an image with the name nginx and the tag ulsah to distinguish it from the official NGINX image. The trailing dot tells `docker build` where to search for the Dockerfile (in this case, the current directory).

Now we can run the image and see our customized `index.html`:

```
# docker run -p 80:80 --name nginx-ulsah -d nginx:ulsah
$ curl localhost
<!DOCTYPE html>
<title>ULSAH index.html file</title>
<p>A simple Docker image, brought to you by ULSAH.</p>
```

We can check that our image is listed among the local images by running the command `docker images`:

```
# docker images | grep ulsah
REPOSITORY  TAG IMAGE ID CREATED   SIZE
nginx        ulsah c0c25eaf7415 3 minutes ago   134.6 MB
```

To remove images, run `docker rmi`. You can’t remove an image until you’ve stopped and removed any containers that are using it:

```
# docker ps | grep nginx:ulsah
IMAGE COMMAND STATUS PORTS
nginx:ulsah "nginx -g 'daemon off'" Up 37 seconds 0.0.0.0:80->80/tcp
# docker stop nginx-ulsah & docker rm nginx-ulsah
nginx-ulsah
nginx-ulsah
# docker rmi nginx:ulsah
```

Both `docker stop` and `docker rm` echo the name of the container they affect, resulting in “nginx-ulsah” being printed twice.
Registries

A registry is an index of Docker images that **dockerd** can access through HTTP. When an image is requested that doesn’t exist on the local disk, **dockerd** pulls it from the registry. Images are uploaded to a registry with **docker push**. Although image operations are initiated by the **docker** command, only **dockerd** actually interacts with registries.

Docker Hub is a hosted registry service run by Docker, Inc. It hosts images for many distributions and open source projects, including all our example Linux systems. The integrity of these official images is verified through a content trust system, thus ensuring that the image you download is provided by the vendor whose name is on the label. You can also publish your own images to Docker Hub for others to use.

Anyone can download public images from Docker Hub, but with a subscription you can also create private repositories. Once you have a paid account at hub.docker.com, log in from the command line with **docker login** to access the private registry so that you can push and pull your own custom images. You can also trigger an image build whenever a commit is detected on a GitHub repository.

Docker Hub is not the only subscription-based registry. Others include quay.io, Artifactory, Google Container Registry, and the Amazon EC2 Container Registry.

Docker Hub is the generous benefactor of the greater image ecosystem, and it also benefits from being the default registry when nothing more specific is requested. For example, the command

```bash
# docker pull debian:jessie
```

first looks for a local copy of the image. If the image isn’t available locally, the next stop is Docker Hub. You can tell **docker** to use a different registry by including a hostname or URL in the image specification:

```bash
# docker pull registry.admin.com/debian:jessie
```

Similarly, when building an image to push to a custom registry, you must tag it with the registry’s URL, and you must authenticate before you push:

```bash
# docker tag debian:jessie registry.admin.com/debian:jessie .
...

# docker login https://registry.admin.com
Username: ben
Password: <password>
# docker push registry.admin.com/debian:jessie
```

Docker saves the login details to a file in your home directory called **.dockerconfig** so that you need not log in again the next time you interact with the private registry.
For performance or security reasons, you might prefer to run your own image registry. The registry project is open source (github.com/docker/distribution), and a simple registry is easy to run as a container:

```
# docker run -d -p 5000:5000 --name registry registry:2
```

The `registry:2` tag differentiates the latest-generation registry from the previous version, which implements an API that is incompatible with current versions of Docker.

After running this command, the registry service is now running on port 5000. You can pull an image from it by qualifying the name of the image you’re seeking:

```
# docker pull localhost:5000/debian:jessie
```

The registry implements two authentication methods: `token` and `htpasswd`. `token` delegates authentication to an external provider, which is likely to require custom development effort. `htpasswd` is simpler and allows HTTP basic authentication for registry access. Alternatively, you can set up a proxy (e.g., NGINX) to handle authentication. Always run the registry with TLS.

The default private registry configuration is not appropriate for a large-scale deployment. Considerations for production use include storage space, authentication and authorization requirements, image cleanup, and other maintenance tasks.

As your containerized environment expands, your registry will be inundated with new images. For users working in the cloud, an object store such as Amazon S3 or Google Cloud Storage is one possible way to store all this data. The registry natively supports both services.

Better yet, you can outsource your registry functions to the registries built into your cloud platform of choice and have one less thing to worry about. Both Google and Amazon run managed container registry services. You pay for storage and for the network traffic to upload and download images.
25.3 Containers in Practice

Once you’re comfortable with the general way that containers work, you’ll find that certain administrative chores need to be approached differently in a containerized world. For example, how do you manage log files for containerized applications? What are some security considerations? How do you troubleshoot errors?

The list below offers a few rules of thumb to help you adjust to life inside a container:

- When your application needs to run a scheduled job, don’t run cron in a container. Use the cron daemon from the host (or a systemd timer) to schedule a short-lived container that runs the job and exits. Containers are meant to be disposable.

- Need to log in and check out what a process is doing? Don’t run sshd in your container. Log in to the host with ssh, then use docker exec to open an interactive shell.

- If possible, set up your software to accept its configuration information from environment variables. You can pass environment variables to containers with the -e KEY=value argument to docker run. Or set up many variables at once from a separate file with --env-file filename.

- Ignore the commonly dispensed advice “one process per container.” That’s nonsense. Split processes into separate containers only when it makes sense to do so. For example, it’s usually a good idea to run an application and its database server in separate containers because they are separated by clear architectural boundaries. But it’s perfectly OK to have more than one process in a container when that’s appropriate. Use common sense.

- Focus on the automatic creation of containers for your environment. Write scripts to build images and upload them to registries. Make sure that software deployment procedures involve replacing containers, not updating them in place.

- On that note, avoid maintaining containers. If you’re accessing a container manually to fix something, figure out what the problem is, resolve it in the image, then replace the container. Immediately update your automation tooling if necessary.

- Stuck? Ask questions on the Docker User mailing list, on the Docker Community Slack, or in the #docker IRC channel on freenode.

Everything an application needs to function should be available within its container: the filesystem, network access, and kernel facilities. The only processes that run in a container are the ones that you start. It is atypical of containers to run normal OS services such as cron, rsyslogd, and sshd, although it is certainly possible to do so. Those duties are best left to the host
OS. If you find yourself needing these services within a container, reconsider your problem and see if you can solve it in a more container-friendly way.
Logging

UNIX and Linux applications traditionally use syslog (now the rsyslogd daemon) to process log messages. Syslog handles log filtering, sorting, and routing to remote systems. Some applications don’t use syslog and instead write directly to log files.

Containers do not run syslog. Instead, Docker collects the logs for you through logging drivers. Container processes need only write logs to STDOUT and errors to STDERR. Docker collects those messages and sends them to a configurable destination.

If your software supports logging only to files, apply the same technique as the NGINX example on this page: create symbolic links from log files to /dev/stdout and /dev/stderr when you build the image.

Docker forwards the log entries it receives to a selectable logging driver. Table 25.4 lists some of the more common and useful logging drivers.

Table 25.4: Docker logging drivers

<table>
<thead>
<tr>
<th>Driver</th>
<th>What it does</th>
</tr>
</thead>
<tbody>
<tr>
<td>json-file</td>
<td>Writes JSON logs in the daemon’s data directory (default)(^a)</td>
</tr>
<tr>
<td>syslog</td>
<td>Writes logs to a configurable syslog destination(^b)</td>
</tr>
<tr>
<td>journald</td>
<td>Writes logs to the systemd journal(^a)</td>
</tr>
<tr>
<td>gelf</td>
<td>Writes logs in the Graylog Extended Log Format</td>
</tr>
<tr>
<td>awslogs</td>
<td>Writes logs to the AWS CloudWatch service</td>
</tr>
<tr>
<td>gcplogs</td>
<td>Writes logs to Google Cloud Logging.</td>
</tr>
<tr>
<td>none</td>
<td>Does not collect logs</td>
</tr>
</tbody>
</table>

\(^a\) Log entries stored this way are accessible through the docker logs command.
\(^b\) Supports UDP, TCP, and TCP+TLS.

When using json-file or journald, you can access log data from the command line through docker logs container-id.

You set the default logging driver for dockerd with the --log-driver option. You can also assign a logging driver at container run time with docker run --logging-driver. Some drivers accept additional options. For example, the --log-opt max-size option configures log file rotation for the json-file driver. Use this option to avoid filling up the disk with log files. Refer to the Docker logging documentation for complete details.
Security advice

Container security relies on processes within containers being unable to access files, processes, and other resources outside their sandbox. Vulnerabilities that allow attackers to escape containers—known as breakout attacks—are serious but rare. The code that underlies container isolation has been in the Linux kernel since at least 2008; it’s mature and stable. As with bare-metal or virtualized systems, insecure configurations are a far more likely source of compromises than are vulnerabilities in the isolation layer.

Docker maintains an interesting list of known software vulnerabilities that are and are not mitigated by containerization. See docs.docker.com/engine/security/non-events.

Restrict access to the daemon

Above all, protect the docker daemon. Because dockerd necessarily runs with elevated privileges, it’s trivial for any user with access to the daemon to gain full root access to the host.

The following sequence of commands demonstrates the risk:

```
$ id
uid=1001(ben) gid=1001(ben) groups=1001(ben),992(docker)
# docker run --rm -v /:/host -t -i debian bash
test@e51ae86c5f7b:/# cd /host
test@e51ae86c5f7b:/host# ls
bin  dev  home  lib64  mnt  proc  run  srv  test  usr
boot  etc  lib  media  opt  root  sbin  sys  tmp  var
```

This transcript shows that any user in the docker group can mount the host’s root filesystem to a container and gain full control of its contents. This is just one of many possible ways to elevate privileges through Docker.

If you use the default UNIX domain socket to communicate with the daemon, add only trusted users to the docker group, which has access to the socket. Better yet, control access through sudo.

Use TLS

See this page for information about TLS.

We said it before, and we’ll say it again: if the docker daemon must be remotely accessible (dockerd -H), require the use of TLS to encrypt network communications and to mutually authenticate the client and server.

Setting up TLS involves having a certificate authority issue certificates to the docker daemon and clients. Once the key pairs and certificate authority are in place, actually enabling TLS for
**docker** and **dockerd** is a simple matter of supplying the right command-line arguments. Table 25.5 lists the essential settings.

**Table 25.5: TLS arguments common to docker and dockerd**

<table>
<thead>
<tr>
<th>Argument</th>
<th>Meaning or argument</th>
</tr>
</thead>
<tbody>
<tr>
<td>--tlsverify</td>
<td>Require authentication</td>
</tr>
<tr>
<td>--tlscert³</td>
<td>Path to a signed certificate</td>
</tr>
<tr>
<td>--tlskeyᵃ</td>
<td>Path to a private key</td>
</tr>
<tr>
<td>--tlscacert³</td>
<td>Path to the certificate of a trusted authority</td>
</tr>
</tbody>
</table>

ᵃ. Optional. The default locations are `~/docker/[cert, key, ca].pem`.

Successful use of TLS relies on a mature certificate management processes. Certificate issuance, revocation, and expiration are a few of the issues that need attention. Heavy is the burden of a security-conscious administrator.

**Run processes as unprivileged users**

Processes in containers should run as nonroot users, just as they should on a full-fledged operating system. This practice limits an attacker’s ability to launch breakout attacks. When you are writing a **Dockerfile**, use the **USER** instruction to run future commands in the image under the named user account.

**Use a read-only root filesystem**

To further restrict containers, you can specify **docker run --read-only**, thereby limiting the container to a read-only root filesystem. This works well for stateless services that never need to write. You can also mount a read/write volume that your process can modify, but leave the root filesystem read-only.

**Limit capabilities**

See [this page](#) for more information about Linux capabilities.

The Linux kernel defines 40 separate capabilities that can be assigned to processes. By default, Docker containers are granted a large subset of these. You can enable an even greater subset by starting a container with the **--privileged** flag. However, this option disables many of the isolation benefits of using Docker. You can tune the specific capabilities that are available to containerized processes with the **--cap-add** and **--cap-drop** arguments:

```
# docker run --cap-drop SETUID --cap-drop SETGID debian:jessie
```

You can also drop all privileges and add back just the ones you need:
Secure images

The Docker content trust feature validates the authenticity and integrity of images in a registry. The publisher of the image signs it with a secret key, and the registry validates it with the corresponding public key. This process ensures that the image was produced by the expected creator. You can use content trust to sign your own images or to validate the images in a remote registry. The feature is available on Docker Hub and on some third party registries, such as Artifactory.

Unfortunately, most of the content on Docker Hub is unsigned and should be considered untrustworthy. Indeed, most images on the Hub are never patched, updated, or audited in any way.

This lack of a proper chain of trust associated with many Docker images is representative of the miserable state of security on the Internet in general. It’s quite common for software packages to depend on third party libraries with little or no concern being given to the trustworthiness of the content that’s pulled in. Some software repositories have no cryptographic signatures whatsoever. It’s also common to find articles that actively encourage disabling validation. Responsible system administrators are highly suspicious of unknown and untrusted software repositories.
**Debugging and troubleshooting**

Containers bring with them a particularly heinous complement of obscure debugging techniques. When an application is containerized, its symptoms become more difficult to characterize and their root causes more puzzling. Many applications can run without modification inside a container, but in some scenarios they may behave differently. You might also encounter bugs within Docker itself. This section helps navigate these treacherous waters.

Errors usually manifest themselves in log files, so that’s the first place to look for information. Use the advice in [Logging](#) to configure logging for containers, and always review the logs when you encounter issues.

If you experience problems with a running container, try

```
docker exec -ti container-name bash
```

to open an interactive shell. From there you can attempt to reproduce the problem, examine the filesystem for evidence, and search for configuration errors.

If you see errors related to the docker daemon or if you have trouble starting it, search the issues list at [github.com/moby/moby](https://github.com/moby/moby). You may find others that have the same problem, and one of them may have identified a potential fix or workaround.

Docker doesn’t automatically clean up images or containers. When neglected, these remnants can consume an inordinate amount of disk space. If your container workload is predictable, configure a **cron** job to clean up by running `docker system prune` and `docker image prune`.

A related annoyance are “dangling” volumes, volumes that were once attached to a container but for which the container has since been removed. Volumes are independent of containers, so any files within them will continue to consume disk space until the volumes are destroyed. You can use the following incantation to clean out orphaned volumes:

```
# docker volume ls -f dangling=true  # List dangling volumes
# docker volume rm $(docker volume ls -qf dangling=true)  # Remove 'em
```

Base images you depend on may have a `VOLUME` instruction in their **Dockerfile**. If you don’t notice this case, you might end up with a full disk after running a few containers from that image. You can show the volumes associated with a container by running `docker inspect`:

```
# docker inspect -f '{{ .Volumes }}' container-name
```
25.4 Container clustering and management

One of the great promises of containerization is the prospect of co-locating many applications on the same host while avoiding interdependencies and conflicts, thereby making more efficient use of servers. This is an appealing vision, but the Docker engine is responsible only for individual containers, not for answering the broader question of how to run many containers on distributed hosts in a highly available configuration.

Configuration management tools such as Chef, Puppet, Ansible, and Salt all support Docker. They can ensure that hosts run a certain set of containers with declared configurations. They also support image building, registry interfaces, network and volume management, and other container-related chores. These tools centralize and standardize container configuration, but they do not solve the problem of conducting the deployment of many containers across a network of servers. (Note that although configuration management systems are useful for a variety of container-related tasks, you will rarely need to use configuration management inside of containers.)

For network-wide container deployments, you need container orchestration software, also known as container scheduling or container management software. An entire symphony of open source and commercial tooling is available to handle large numbers of containers. Such tools are crucial for running containers at scale in a production context.

To understand how these systems work, think of the servers on the network as a farm of compute capacity. Each node in the farm offers CPU, memory, disk, and network resources to the scheduler. When the scheduler receives a request to run a container (or set of containers), it places the container on a node that has sufficient spare resources to meet the container’s needs. Because the scheduler knows where containers have been placed, it can also assist in routing network requests to the correct nodes within the cluster. Administrators interact with the container management system rather than dealing with any individual container engine. Exhibit D illustrates this architecture.

Exhibit D: Basic container scheduler architecture
Container management systems supply several helpful features:

- Scheduling algorithms select the best node in light of a job’s requested resources and the utilization of the cluster. For example, a job with high bandwidth requirements might be slotted onto a node with a 10 Gb/s network interface.

- Formal APIs allow programs to submit jobs to the cluster, opening the door to integration with external tools. It’s easy to use container management systems in conjunction with CI/CD systems to simplify software deployments.

- Container placement can accommodate the needs of high-availability configurations. For example, an application may need to run on host nodes in several distinct geographical regions.

- Health monitoring is built in. The system can terminate and reschedule unhealthy jobs and can route jobs away from unhealthy nodes.

- It’s easy to add or remove capacity. If your compute farm doesn’t have enough resources available to satisfy demand, you can simply add another node. This facility is especially well suited to cloud environments.

- The container management system can interface with a load balancer to route network traffic from external clients. This facility obviates the complex administrative process of manually configuring network access to containerized applications.

One of the most challenging tasks in a distributed container system is mapping service names to containers. Remember that containers are typically ephemeral in nature and may have dynamic ports assigned. How do you map a friendly, persistent service name to multiple containers, especially when the nodes and ports change frequently? This problem is known as service discovery, and container management systems have various solutions.
It helps to be familiar with the underlying container execution engine before diving into orchestration tooling. All the container management systems we’re aware of rely on Docker as the default container execution engine, although some systems also support other engines.
A synopsis of container management software

Despite their relative youth, the container management tools we discuss below are mature beyond their years and can be considered production grade. In fact, many are already used in production at high-profile, large-scale technology companies. Most are open source and have sizable user communities. Based on recent trends, we anticipate substantial development in this area in the coming years.

In the upcoming sections, we highlight the functionality and features of the most widely used systems. We also mention their integration points and common use cases.
**Kubernetes**

Kubernetes—sometimes shortened to “k8s” because there are eight letters between the leading “k” and the trailing “s”—has emerged as a leader in the container management space. It originated within Google and was launched by some of the same developers that worked on Borg, Google’s internal cluster manager. Kubernetes was released as an open source project in 2014 and now has more than a thousand active contributors. It has the most features and the fastest development cycle of any system we’re aware of.

Kubernetes consists of a few separate services that integrate to form a cluster. The basic building blocks include

- The API server, for operator requests
- A scheduler, for placing tasks
- A controller manager, for tracking the state of the cluster
- The Kubelet, an agent that runs on all cluster nodes
- cAdvisor, for monitoring container metrics
- A proxy, for routing incoming requests to the appropriate container

The first three items on this list run on a set of masters (which can optionally serve dual duty as nodes) for high availability. The Kubelet and cAdvisor processes run on each node, handling requests from the controller manager and reporting statistics about the health of their tasks.

In Kubernetes, containers are deployed as a “pod” which contains one or more containers. All containers in a pod are guaranteed to be co-located on the same node. Pods are assigned a cluster-wide unique IP address, and they are labeled for identification and placement purposes.

Pods are not meant to be long-lived. If a node dies, the controller schedules a replacement pod on a different node with a new IP address. Therefore, you cannot use the address of a pod as a durable name.

Services are collections of related pods with an address that is guaranteed not to change. If a pod within a service dies or fails a health check, the service removes that pod from its rotation. You can also use the built-in DNS server to assign resolvable names to services.

Kubernetes has integrated support for service discovery, secret management, deployment, and pod autoscaling. It has pluggable networking options to enable container network overlays. It can support stateful applications by migrating volumes among nodes as needed. Its CLI tool, `kubectl`, is one of the most intuitive that we’ve ever worked with. In short, it has more advanced features than we can possibly cover in this short section.
Although Kubernetes has the most active and engaged community and the most advanced features, those assets are accompanied by a steep learning curve. Recent versions have improved the experience for first-time users, but a full-fledged, customized Kubernetes deployment is not for the timid. Production k8s deployments impose a substantial administrative and operational burden.

The Google Container Engine service is implemented with Kubernetes, and it offers one of the best experiences for teams that want to run containerized workloads without the operational overhead of cluster management.
Mesos and Marathon

Mesos is an entirely different breed. It was conceived at the University of California at Berkeley around 2009 as a generic cluster manager. It quickly made its way to Twitter, where it now runs on thousands of nodes. Today, Mesos is a top-level project from the Apache Foundation and boasts a large number of enterprise users.

The major conceptual entities in Mesos are masters, agents, and frameworks. A master is a proxy between agents and frameworks. Masters relay offers of system resources from agents to frameworks. If a framework has a task to run, it chooses an offer and instructs the master to run the task. The master sends along the task details to the agent.

Marathon is a Mesos framework that deploys and manages containers. It includes a handsome user interface for managing applications and a simple, RESTful API. To run an application, you write a request definition in JSON format and submit it to Marathon through the API or the UI. Because it’s an external framework, the deployment of Marathon is flexible. Marathon can run on the same node as the master for convenience, or it can run externally.

Support for multiple, coexisting frameworks is Mesos’s biggest differentiator. Apache Spark, the big-data processing tool, and Apache Cassandra, a NoSQL database, both offer Mesos frameworks, thus allowing you to use Mesos agents as nodes in a Spark or Cassandra cluster. Chronos is a framework for scheduled jobs, rather like a version of cron that runs on a cluster instead of an individual machine. The ability to run so many frameworks on the same set of nodes is a nice feature and helps create a unified and centralized experience for administrators.

Unlike Kubernetes, Mesos does not come with batteries included. For example, load balancing and traffic routing are pluggable options that depend on your preferred solution. Marathon includes a tool, the Marathon-lb, that implements this service, or you can choose your own. We’ve had success using HashiCorp’s Consul and HAProxy. Designing and implementing an exact solution is left as an exercise for the administrator.

Like Kubernetes, Mesos requires some contemplation to understand and use. Mesos and most of its frameworks rely on Apache Zookeeper for cluster coordination. Zookeeper is somewhat difficult to administer and is known for complex failure cases. In addition, a high-availability Mesos cluster requires a minimum of three nodes, which may be an onerous burden at some sites.
Docker Swarm

Not to be left behind, Docker offers Swarm, a container cluster manager built directly into Docker. The current incarnation of Swarm emerged in 2016 as an answer to the growing popularity of Mesos, Kubernetes, and other cluster managers that used Docker containers under the hood. Container orchestration is now a major focus for Docker, Inc.

Swarm is easier to get started with than is Mesos or Kubernetes. Any node that runs Docker can join the swarm as a worker node, and any worker node can also be a manager. There is no need to run separate nodes as masters. (Strictly speaking, this is true for Kubernetes and Mesos as well, but we’ve found it to be common practice to separate masters from agents in high-availability configurations.)

Starting a swarm is as simple as running `docker swarm init`. There are no additional processes to manage and configure, and there is no state to track. It works out of the box.

You can use familiar `docker` commands to run services (as in Kubernetes, collections of containers) on the swarm. You declare the state you want to achieve (“three containers running my web application”) and Swarm schedules the tasks on the cluster. It automatically handles failure states and zero-downtime updates.

Swarm has a built-in load balancer that adjusts automatically as containers are added or removed. The Swarm load balancer is not as full-featured as tools such as NGINX or HAProxy, but on the other hand, it doesn’t require any administrative attention.

Swarm supplies a secure Docker experience by default. All connections between nodes in a swarm are TLS-encrypted, and no configuration is required on the part of the administrator. This is a major differentiator for Swarm when compared to its competitors.
AWS EC2 Container Service

AWS offers ECS, a container management service designed for EC2 instances (AWS’s native virtual servers). In a manner reminiscent of many Amazon services, AWS launched ECS with minimal functionality but has steadily enhanced it over time. ECS has matured into a fine choice for sites that are already invested in AWS and want to stick to E-Z mode.

ECS is a “mostly managed” service. The cluster manager components are operated by AWS. Users run EC2 instances that have Docker and the ECS agent installed. The agent connects to the central ECS API and registers its resource availability. To run a task on your ECS cluster, you submit a task definition in JSON format through the API. ECS then schedules the task on one of your nodes.

Because the service is mostly managed, the barrier to entry is low. You can get started with ECS in just a few minutes. The service scales well to at least hundreds of nodes and thousands of concurrent tasks.

ECS integrates with other AWS services. For example, load balancing among multiple tasks, along with the requisite service discovery, are handled by the Application Load Balancer service. You can add resource capacity to your ECS cluster by taking advantage of EC2 autoscaling. ECS also integrates with AWS’s Identity and Access Manager service to grant permissions for your container tasks to interact with other services.

One of the most polished parts of ECS is the included Docker image registry. You can upload Docker images to the EC2 Container Registry, where they’re stored and made available to any Docker client, whether it’s running on ECS or not. If you’re running containers on AWS, use the container registry in the same region as your instances. You’ll achieve far better reliability and performance than with any other registry.

The ECS user interface, although functional, shares the limitations of other AWS interfaces. The AWS CLI tool has complete support for the ECS API. For management of applications on ECS, we recommend turning to third party, open source tools such as Empire (github.com/remind101/empire) or Convox (convex.com) for a more streamlined experience.
25.5 RECOMMENDED READING

**DOCKER, INC. Official Docker Documentation.** docs.docker.com. Docker has good documentation. It’s comprehensive and usually up to date.


**MOUAT, ADRIAN. Using Docker: Developing and Deploying software with Containers.** Sebastopol, CA: O’Reilly Media, 2016. This book covers topics from basic to advanced and includes plenty of examples.


The Container Solutions blog at container-solutions.com/blog includes technical HOWTOs, best practices, and interviews with experts in the container space.
Until the past decade or so, updating software was a hair-pulling, time-consuming exercise in frustration. Release processes typically involved ad hoc, home-grown scripts that were invoked in enigmatic order and saddled with outdated and incomplete documentation. Testing—if it existed at all—was performed by a quality assurance team that was far removed from the development cycle and often became a major obstacle to shipping code. Administrators, developers, and project managers would plan days-long marathons for the final stages of releasing updates to live users. Service outages were often scheduled weeks in advance.

Given this unsavory context, it should come as no surprise that some very smart people were working diligently to improve the situation. After all, where some see only problems, others see opportunity.

At top of mind is Martin Fowler, an oracle of the software industry and chief scientist of the influential development shop ThoughtWorks. In an insightful article (goo.gl/Y2lisI), Fowler describes continuous integration as “a software development practice where members of a team integrate their work frequently,” thus removing one of the major pain points of software work: the tiresome task of reconciling code fragments that have diverged dramatically over a long period of independent development. The practice of continuous integration is now ubiquitous among software development teams.
Hot on the heels of this innovation came continuous delivery, which is similar in concept but which targets a separate goal: reliably deploying updated software to live systems. Continuous delivery embraces the release of small, incremental changes to IT infrastructure. If something breaks (that is, if a “regression” is introduced), it becomes straightforward to isolate and resolve the issue because the changes between versions are small. At the extreme end, some sites aim to deploy new code to users multiple times per day. Bugs and security issues can be resolved in hours rather than weeks.

In combination, continuous integration and continuous delivery (henceforth denoted CI/CD) encompass the tools and processes needed to facilitate frequent, incremental software and configuration updates.

See this page for more comments on DevOps.

CI/CD is a pillar of the DevOps philosophy. It’s the glue that holds together developers and operations folks. It is as much a business asset as a technical innovation. Once introduced, CI/CD becomes the bedrock of an IT organization because it imposes logic and organization on release processes that were previously chaotic.

Sysadmins are central to the design, implementation, and ongoing maintenance of CI/CD systems. Administrators install, configure, and operate the tools that make CI/CD function. They are responsible for ensuring that software build processes are fast and reliable.

Testing is an important element of CI/CD, and although administrators may not write the tests (though they sometimes do!), they are often responsible for setting up the infrastructure and the systems on which the tests are performed. Perhaps most importantly, it is ultimately system administrators who are responsible for deployments, the “delivery” component of CI/CD.

An effective CI/CD system is implemented not with a solitary tool but rather with a collection of software that works in unison to form a cohesive environment. Myriad open source and commercial tools are available to coordinate the various elements of CI/CD. These coordination tools rely on other software packages to do the actual work (e.g., compiling code or setting up servers in a particular configuration). Indeed, there are so many options that the initial approach to CI/CD can be overwhelming. If nothing else, the recent proliferation of tools in this space is evidence of CI/CD’s growing importance to the industry.

In this chapter we attempt to navigate the maze of CI/CD concepts, terminology, and tools. We cover the basics of a CI/CD pipeline, the various types of testing and their relevance to CI/CD, the practice of running multiple environments in parallel, and some of the most popular open source tools. At the end of the chapter, we dissect an example CI/CD pipeline that uses some of the most popular tools. When you’re finished with this chapter, you should understand some of the principles and techniques that go into creating a powerful, flexible CI/CD system.
26.1 CI/CD ESSENTIALS

Many terms related to CI/CD sound similar and have overlapping meanings. So let’s first take a closer look at the differences between continuous integration, delivery, and deployment:

- **Continuous integration** is the process of collaborating on a shared code base, merging disparate code changes into a version control system, and automatically creating and testing builds.

- **Continuous delivery** is the process of automatically deploying builds to nonproduction environments after the continuous integration process completes.

- **Continuous deployment** closes the loop by deploying to live systems that serve real users without any operator intervention.

Continuous deployment without any human supervision can be intimidating, but that’s precisely the point: the idea is to reduce the fear factor by deploying as often as possible, eliminating more and more issues until the team has enough confidence in the testing and tooling to enable automatic releases.

Continuous deployment need not be the ultimate goal of all sites. There may be compliance or risk reasons to pause at any point in the pipeline. If that’s the case, you can still benefit from making each stage of the process as simple as possible for the human who pushes the final button. Every organization should choose its own boundaries.
Principles and practices

Business agility is one of the key benefits of CI/CD. Continuous deployment facilitates the release of well-tested features to production in minutes or hours instead of weeks or months. Because every change is built, tested, and deployed immediately, the delta between versions is much smaller. And that decreases the risk of deployment and helps narrow the range of possible root causes if something goes wrong. Rather than staging a small number of big-bang deployments per year, you might find yourself releasing new code multiple times per week or even per day.

CI/CD stresses the release of more features, more often. This goal is achievable only when developers write and commit code in smaller chunks. To realize continuous integration, developers need to push code changes at least once per day after running tests locally.

For administrators, CI/CD processes greatly reduce the amount of time spent preparing for and implementing releases. They also reduce the time spent debugging problems when deployments inevitably fail. Few things are more satisfying than watching a new feature release itself to production without any human intervention.

The next sections cover some basic rules of thumb to keep in mind as you develop your CI/CD processes.

Use revision control

All code should be tracked in a source control system. We recommend Git, but there are lots of options. Most software development teams use source control as a matter of course.

For sites that embrace the infrastructure-as-code concept (as we demonstrate in the section CI/CD in practice), you can track infrastructure-related code alongside your applications. You can even store documentation and configuration settings in source control.

Make sure version control is the single source of truth. Nothing can be managed manually or off the record.

Build once, deploy often

A CI/CD pipeline begins with a build. The output of the build (the “artifact”) is used from that point forward for testing and deployment. The only way to confirm that a specific build is ready to go to production is to run all tests against that build. Deploy the same artifact to at least one or two environments that match production as closely as possible.

Automate end-to-end

Building, testing, and deploying code without manual intervention is the key to reliable and reproducible updates. Even if you’re not planning to deploy code continuously to production, the final production deployment step should run fully unattended after being triggered by a human.
Build every integration commit

An integration merges changes made by multiple developers or teams of developers. The product is a composite code base that incorporates everyone’s updates.

An integration does not randomly snatch work in progress out of developers’ hands and stick it in the mainline code base; that’s a recipe for disaster. Individual developers are responsible for managing their own development stream. When they’re ready, they initiate an integration. Integrations occur as frequently as possible.

Integrations are performed through the source control system. The exact workflow varies. Individual developers might be responsible for merging their work back to the trunk, or a designated release overseer might integrate several developers or teams at once. The merge process can be largely automated, but there’s always the potential for two sets of changes to conflict. That situation requires human intervention.

The idea behind continuous integration is that commits to the revision control system’s integration branch automatically result in a build. The “integration branch” part is important because source control serves several purposes. In addition to being a vehicle for collaboration and integration, it’s also useful as a backup system, as a checkpoint for work in progress, and as a system that lets developers work on several updates while keeping the changes related to those updates logically separate. Therefore, only commits to the integration branch result in a build.

Frequent integrations make it easy to trace a broken build back to the exact lines of code that caused the problem. The revision control system can then determine the identity of the responsible developer. But note: a broken build should carry little or no stigma. The goal is just to get the build running again. Encourage a blame-free culture within your teams.

Share responsibility

When something goes wrong, the pipeline needs to be fixed. No new code can be pushed until the previous problem has been resolved. It’s the equivalent of halting the assembly line in a factory. It is the responsibility of the entire team to fix the build before resuming development work.

CI/CD shouldn’t be a mysterious system that runs in the background and occasionally sends email when something is broken. Every team member should have access to the CI/CD interface to view dashboards and logs. Some sites create humorous widgets such as RGB lighting fixtures that act as a visual indicator of the pipeline’s current status.

Build fast, fix fast

CI/CD is designed to yield feedback as quickly as possible, ideally within minutes after pushing code to source control. This rapid response guarantees that developers pay attention to the result. If the build fails, the developers will likely be able to fix the problem quickly because the changes they just committed are fresh in their minds. Slow build processes are
counterproductive. Strive to eliminate redundant and time-consuming steps. Ensure that your build system has enough agents, and that the agents have sufficient system resources to build quickly.

Audit and verify

Part of the CI/CD system includes a detailed history of every software release, including its progression from development to production. This auditability can be useful to ensure that only authorized builds are deployed. The settings and event timelines related to each environment can be irrefutably verified.
Environments

Applications do not run in isolation. They depend on external resources such as databases, caches, network filesystems, DNS records, remote HTTP APIs, other applications, and external network services. An execution environment includes all these resources and anything else that the application needs so it can run. Building and maintaining such environments is a target of substantial administrative attention.

Most sites run at least three environments, listed here in ascending order of importance:

- Development ("dev" for short), for integrating updates from multiple developers, testing infrastructure changes, and checking for obvious failures. Development is used mostly by the technical staff and not by business types or end users. In the context of CI/CD, the development environment may be created and reset multiple times per day.

- Staging (or "stage"), for manual and automated testing and for further vetting of changes and software updates. Some organizations call this the "test" environment. Testers, product owners, and other business stakeholders use the staging environment to review new features and bug fixes. Staging can also be used for penetration testing and other security checks.

- Production ("prod"), for implementing service for real users. The production environment usually includes extensive measures to ensure high performance and strong security. An outage in production is an all-hands-on-deck emergency that must be resolved immediately.

A typical CI/CD system promotes software through each of these environments in succession, filtering out errors and software defects along the way. You can deploy to production with confidence because you know that changes have already been tested in two other environments.

Environment parity is a subject of some complexity for administrators. The purpose of the nonproduction or "lower" environments is to prepare and scrutinize changes of all types before they are made in production. Substantive differences among environments can result in unforeseen incompatibilities that might ultimately cause degraded performance, downtime, or even the destruction of data.

For example, imagine that the development and staging environments have undergone an operating system upgrade, but production still runs the older OS version. It’s time for a software deployment. The new software is thoroughly tested in dev and stage, and it seems to work fine. However, an unexpected incompatibility becomes evident during the production rollout because the older version of a certain library is missing functionality used by the new code.

This scenario is quite common, and it’s one reason why administrators must be vigilant about keeping environments in sync. The closer that lower environments match production, the higher
your chances of maintaining high availability and delivering software successfully.

Running multiple environments at full capacity can be expensive and time-consuming. Because production serves far more users than the lower environments, it’s usually necessary to run a larger number of more expensive systems in that environment. Production data sets tend to be larger, and the provisioned disk space and server size are beefed up to compensate.

Even this type of difference among environments can cause unanticipated problems. A load balancer misconfiguration that didn’t matter in dev or stage may reveal a defect, or a database query that runs quickly in dev and stage might turn out to be far slower when applied to production-scale data.

Matching production capacity in lower environments is a tricky problem. Strive to have at least one lower environment that has redundancy in all the same places that production does (e.g., multiple web servers, fully replicated databases, and matching failover strategies for any clustered systems). It’s fine for the staging servers to be smaller in size, although any tests you run to check performance will not reflect production numbers.

For best results, data sets in lower environments should be similar in size and content to those of production. One strategy is to create nightly snapshots of all relevant production data and copy it to the lower environment. For compliance and good security hygiene, sensitive user data must be anonymized before it’s used this way. For truly massive data sets that are not practical to copy, import a smaller but still meaningful sample.

Despite your best efforts, lower environments will never be exactly like the production environment. Some configuration settings (such as credentials, URLs, addresses, and hostnames) will differ. Use configuration management to track these configuration items among environments. When the CI/CD system runs a deployment, consult your source of truth to find the relevant configuration for that environment and make sure that all environments are deployed in the same way.
**Feature flags**

A feature flag enables or disables an application feature depending on the value of a configuration setting. Developers can build support for feature flags into their software. You can use feature flags to enable certain features in specific environments. For example, you can enable a feature for the staging environment while keeping it disabled in production until it’s fully tested and ready for the user base.

For instance, consider an e-commerce application that has a shopping cart. The business wants to run a promotion that requires some changes to the code. The development team can build the feature and release it to all three environments in advance, but enable it only on dev and stage. When the business is ready to advertise and activate the promotion, enabling the feature becomes a simple, low-risk configuration change rather than a software release. If the feature has a bug, it’s easy to disable it without updating the software.
26.2 Pipelines

A CI/CD pipeline is a series of steps, called “stages,” that run in sequence. Each stage is essentially a script that performs tasks specific to your software project.

At the most basic level, a CI/CD pipeline

- Reliably builds and packages software
- Runs a series of automated tests to search for bugs and configuration errors
- Deploys code to one or more environments, culminating in production

Exhibit A illustrates the stages in a simple (yet mature) CI/CD pipeline.

Exhibit A: A basic CI/CD pipeline

The following sections break down the three stages in further detail.
The build process

A build is a snapshot of the current status of a software project. It’s typically the first stage of any CI/CD pipeline, possibly after a code analysis stage that monitors code quality and searches for security risks. The build step transforms the code into an installable piece of software. Builds can be triggered by a commit to the integration branch of the code repository or they can run on a regular schedule or on demand.

Every pipeline run starts with a build, but not every build reaches production. Once a build passes testing, it becomes a “release candidate.” If the release candidate is actually deployed to production, it becomes a “release.” If you do continuous deployment, every release candidate is also a release. Exhibit B illustrates these categories.

Exhibit B: Builds, release candidates, and releases

<table>
<thead>
<tr>
<th>Builds</th>
<th>Release candidates</th>
<th>Releases</th>
</tr>
</thead>
<tbody>
<tr>
<td>Every time you change code</td>
<td>All tests passed</td>
<td>Deployed to production</td>
</tr>
</tbody>
</table>

The precise steps of the build process depend on the language and software. For a program in C, C++, or Go, the build process is a compilation, often initiated by make, that results in an executable binary. For languages that do not require compilation, such as Python or Ruby, the build stage might involve packaging the project with all relevant dependencies and assets, including libraries, images, templates, and markup files. Some builds might involve only configuration changes.

The output of the build stage is a “build artifact.” The nature of that artifact depends on the software and the configuration of the rest of the pipeline. Table 26.1 lists some of the common types of artifacts. Whatever the format, the artifact is the basis for deployments throughout the rest of the pipeline.

Table 26.1: Common types of build artifacts
Build artifacts are saved to an artifact repository. The type of repository depends on the type of artifact. At its simplest, a repository can be a directory on a remote server that’s accessible through SFTP or NFS. It can also be a yum or APT repository, a Docker image repository, or, in the cloud, an object store such as an AWS S3 bucket. The repository must be available to all the systems that need to download and install the artifact during a deployment.

<table>
<thead>
<tr>
<th>Type</th>
<th>What it’s for</th>
</tr>
</thead>
<tbody>
<tr>
<td>.jar or .war file</td>
<td>Java archive or Java web application archive</td>
</tr>
<tr>
<td>Static binary</td>
<td>Statically compiled programs, commonly C or Go</td>
</tr>
<tr>
<td>.rpm or .deb file</td>
<td>OS-native software packages for Red Hat or Debian</td>
</tr>
<tr>
<td>pip or gem package</td>
<td>Packaged Python or Ruby applications</td>
</tr>
<tr>
<td>Container image</td>
<td>Applications that run under Docker</td>
</tr>
<tr>
<td>Machine image</td>
<td>Virtual servers, especially for public or private clouds</td>
</tr>
<tr>
<td>.exe file</td>
<td>Windows executable</td>
</tr>
</tbody>
</table>
**Testing**

Each stage in a CI/CD pipeline runs tests to catch buggy code and bad builds so that the code that makes it through to production is free of defects (or at least, as free as possible). Testing is the linchpin of this process. It engenders trust that a release is ready to deploy.

If a build fails any test, the remaining stages of the pipeline are pointless. The team must determine why the build failed and address the underlying issue. Because builds are created for every code push, it’s easy to isolate the problem to the latest commit. The fewer lines of code changed between builds, the easier the problem’s isolation.

Failures do not always stem from software bugs. They can occur because of network conditions or infrastructure errors that require administrative attention. If the application depends on outside resources, such as third party APIs, there can be upstream failures in the external resource. Some tests can run in isolation, but other tests require the same infrastructure and data that will be present in production.

Consider adding each of the following types of tests to your CI/CD pipeline:

- **Static code analysis** examines code for syntax errors, duplication, violations of coding guidelines, security problems, or excessive code complexity. These checks are fast and do not involve executing the actual code.

- **Unit tests** are written by the same developers who write the application code. They reflect the developer’s view of how the code is supposed to function. The intent is to test the input and output of every method and function (unit) in the code. “Code coverage” is a (sometimes misleading) metric that describes what portion of the code is being unit-tested.

- **Integration tests** take unit tests one step further by running the application in its intended execution environment. Integration tests run the application with its underlying frameworks and with external dependencies such as outside APIs, databases, queues, and caches.

- **Acceptance tests** simulate typical use. In contrast to unit tests, acceptance tests reflect a user’s point of view. For web-based software, this stage might include remote-controlling browser page loads through tools such as Selenium. For mobile software, the build artifact might go to a device farm that runs the app on many different mobile devices. Different browsers and versions make acceptance tests challenging to create, but in the end, these tests have meaningful results.

- **Performance tests** search for performance problems introduced by the latest code. To identify bottlenecks, these so-called stress tests should invoke the application within a perfect clone of your production environment, with real traffic patterns. Tools such as JMeter or Gatling can simulate thousands of concurrent users interacting with an
application in a programmed pattern. To gain the most from performance testing, ensure that monitoring and graphing instrumentation are in place. Those tools clarify both the application’s typical performance and its behavior under a new build.

- **Infrastructure tests** go hand-in-hand with programmatically provisioned cloud infrastructure. If you create temporary cloud infrastructure as part of your CI/CD pipeline, you can write test cases to verify the proper configuration and operation of the infrastructure itself. Does the system run through configuration management successfully? Are only the expected daemons running? Serverspec (serverspec.org) is one interesting tool in this space.

Depending on the characteristics of your project, some tests are more important than others. For example, software that implements a REST API has no need for browser-based acceptance tests. Instead, you’ll likely focus on integration tests. On the other hand, for shopping cart software, browser tests for all the important user paths (catalog, product pages, cart, checkout) are mandatory. Consider the needs of your project and implement testing accordingly.

Sometimes the code that’s difficult to test is also the most likely to have defects. Your code may have 85% code coverage through unit tests (which is quite high by industry standards), but if the most complex code isn’t tested, bugs might be missed. Code coverage alone is not an adequate measure of code quality.

Testing workflows don’t have to be linear. Actually, because one of the goals is to get feedback as quickly as possible, it’s a good idea to run as much of the testing as possible in parallel. But keep in mind that some tests might depend on the results of other tests; others might potentially interfere with each other. (Ideally, tests should have no cross-dependencies.)

Avoid the temptation to ignore or overlook broken tests. It’s easy to get in the habit of understanding the reason for a failure, considering it to be harmless or inapplicable, and suppressing the test. However, this thinking is dangerous and can lead to a less reliable testing system. Keep in mind the golden rule of CI/CD: fixing a broken pipeline is the top priority.

To reinforce this tenet, make it difficult to ignore failed tests. It should be a technical requirement, enforced through the CI/CD software, that production deployments cannot occur if there are any broken tests.
Deployment

Deployment is the act of installing software and preparing it for use within a server environment. The specifics of how this is done depend on the technology stack. A deployment system must understand how to retrieve the build artifact (e.g., from a package repository or container image registry), how to install it on the server, and what setup steps, if any, are necessary. A deployment concludes when a new version of software is running and the old version has been disabled.

A deployment might be as simple as updating some HTML files on disk. No restart or further configuration required! But for most cases, a deployment involves at least installing a package and restarting an application. Complex, large-scale production deployments might involve installing code on multiple systems while serving live traffic, without pausing for a service outage.

System administrators play an important role in the deployment process. They are usually responsible for creating deployment scripts, monitoring important application health indicators during deployments, and ensuring that the infrastructure and configuration needs of other team members are met.

The following list describes just a few of the possible ways to deploy software:

- Run a basic shell script that invokes `ssh` to log in to each system, downloads and installs the build artifact, and then restarts the application. These types of scripts are usually home grown and do not scale to more than a handful of systems.

- Use a configuration management tool to orchestrate the installation process across a managed set of systems. This strategy is more organized and scalable than the use of shell scripts. Most configuration management systems are not designed specifically to facilitate deployments, although they can be used for this purpose.

- If the build artifact is a container image and the application runs on a container management platform such as Kubernetes, Docker Swarm, or AWS ECS, the deployment might be nothing more than a quick API call to the container manager. The container service manages the rest of the deployment process on its own. See *Containers and CI/CD*.

- A few open source projects standardize and streamline deployment. Capistrano (capistranorb.com) is a Ruby-based deployment tool that extends Ruby’s Rake system to run commands on remote systems. Fabric (fabfile.org) is a similar tool written in Python. These tools, by developers for developers, are essentially elaborate shell scripts.

- Software deployment is a well-explored problem for users of public clouds. Most cloud ecosystems include both integrated and third party deployment services that
can be used from a CI/CD pipeline. Some examples include Google Deployment Manager, AWS CodeDeploy, and Heroku.

Tailor the deployment technique to your site’s technology stack and service needs. If you have a simple environment with a few servers and a small handful of applications, a configuration management tool might be appropriate. At sites with a large number of servers spread among data centers, a specialized deployment tool is called for.

An “immutable” deployment codifies the principle that servers should never be modified (or “mutated”) once they’ve been initialized. To deploy a new release, the CI/CD tooling creates entirely new servers with the updated build artifact included in the image. In this model, servers are considered disposable and temporary. This strategy is predicated on a programmable infrastructure such as a public or private cloud, where instances can be allocated through an API call. Some of the largest users of the public cloud embrace immutable deployments.

In *CI/CD in practice*, we walk through a sample immutable deployment that uses HashiCorp’s Terraform tool to create and update infrastructure.
Zero-downtime deployment techniques

At some sites, services must continue to run even while they are being upgraded or redeployed, either because an outage poses unacceptable risk (health care, government services) or because it might have substantial financial costs (high-volume e-commerce or financial services). Updating live software without service interruptions is the Xanadu of software deployments, and it’s the source of much anxiety and yak-shaving.

One common way to achieve a zero-downtime release is a “blue/green” deployment. The concept is straightforward: stage the new software on a standby system (or set of systems), run tests to confirm its functionality, then switch traffic from the live system to the standby once the tests are complete.

See this page for more information about load balancers.

This strategy works particularly well when traffic is proxied through a load balancer. The live systems handle all the user connections while the standby systems are being prepared. When the time is right, the standby systems can be added to the load balancer and the previously live systems removed. The deployment is complete when all the old systems are out of the rotation and all the transactions they were handling have concluded.

A “rolling” deployment updates existing systems in a stepwise fashion, modifying software on one system at a time. Each system is removed from the load balancer, updated, then added back to the rotation to accept user traffic. This type of deployment can cause problems if the application cannot tolerate two different versions running simultaneously.

Both the blue/green and rolling deployment strategies can accommodate a “canary,” akin to the hapless canary in a coal mine. You first allocate a small amount of traffic to a single system (or small percentage of systems) that runs the new release. If the new release has problems, you back it out and correct the problem, having impacted only a handful of users. Of course, the canary systems need precise telemetry and monitoring so that you can determine whether problems have been introduced.
26.3 **JENKINS: THE OPEN SOURCE AUTOMATION SERVER**

Jenkins is an automation server written in Java. It’s by far the most popular software used to implement CI/CD. With broad adoption and an extensive ecosystem of plug-ins, Jenkins is well suited to a variety of use cases.

It’s easy to dabble with Jenkins by running it in a Docker container:

```bash
$ docker run -p 8080:8080 --name jenkins jenkinsci/jenkins
```

Once the container starts, you can access the Jenkins user interface in a web browser on port 8080. You’ll find the initial administrator password buried in the container output. In practice, you would need to change that password immediately!

A single-container configuration is fine for learning the ropes, but you’ll likely need a more robust solution in production environments. The Jenkins download page ([jenkins.io/download](https://jenkins.io/download)) has installation instructions that we needn’t rehash here. Refer to those docs for installation on Linux and FreeBSD. CloudBees, the maker of Jenkins, also offers a high-availability version called Jenkins Enterprise.

Jenkins has plug-ins for every conceivable task. Use plug-ins to outsource builds to different types of agents, send notifications, coordinate releases, and execute scheduled jobs. Plug-ins integrate with open source tools and with all the major cloud platforms and external SaaS providers. Plug-ins give Jenkins superpowers.

Most Jenkins configuration is done through the web UI, and being merciful to your attention span, we don’t attempt to cover the UI’s nooks and crannies. Instead, we introduce the fundamentals of Jenkins along with some of its most important features.
Basic Jenkins concepts

At its core, Jenkins is a coordination server that links a series of tools into a chain—or, to use CI/CD terminology, a pipeline. Jenkins is an organizer and facilitator; all actual work depends on outside services such as source code repositories, compilers, build tools, testing harnesses, and deployment systems.

A Jenkins job, or project, is a collection of linked stages. Creating a project is the first order of business for a new installation. You can link the project’s steps together so that they run in sequence or in parallel. You can even set up conditional steps that do different things depending on the results of previous steps.

Every project should be connected to a source code repository. Jenkins has native support for pretty much every version control system: Git, Subversion, Mercurial, even ancient systems such as CVS. There are also integration plug-ins for higher-level version control services such as GitHub, GitLab, and BitBucket. You’ll need to give Jenkins the appropriate credentials to allow it to download code from your repository.

The “build context” is the current working directory on the Jenkins system that’s executing a build. Source code is copied into the build context along with any supporting files that are needed for the build.

Once you’ve wired up Jenkins to a version control repository, you can create a build trigger. This is the signal for Jenkins to copy the current source code and start the build process. Jenkins can poll the source repository for new commits and initiate a build whenever it finds one. It can also start builds on a schedule or be triggered by a web hook, a feature supported by GitHub.

After setting up the trigger, create the build steps, that is, the specific tasks that will create a build. Steps can be code-base-specific, or they can be generic shell scripts. For example, Java projects are usually built with a tool called Maven. A Jenkins plug-in supports Maven directly, so you can simply add a Maven build step. For a project written in C, the first build step might just be a shell script that runs `make`.

The remaining build steps depend on your goals for the project. The most common builds include steps that initiate the testing tasks discussed in Testing. You may need a step to create a custom build artifact such as a tarball, OS package, or container image. You can also include steps that trigger administrator notifications, take deployment-related actions, or coordinate with outside tooling.

For a CI/CD project, the build steps can address all the stages of a pipeline: build the code, run tests, upload the artifact to a repository, and kick off a deployment. Each stage of the pipeline is a just a build step within the Jenkins project. The Jenkins interface presents an overview of the status of each step, so it’s easy to see at a glance what’s happening in the pipeline.
Sites that have many applications should have separate Jenkins projects for each. Each project will have a distinct code repository and build steps. The Jenkins system needs all the tools and dependencies to run a build for any of its projects. For example, if you have configured both a Java project and a C project, your Jenkins system must have both Maven and **make** installed.

Projects can depend on other projects. Use this to your advantage by structuring projects as generic, inheritable templates. For example, if you have a variety of applications that are built differently but deployed in the same way (e.g., as containers running on a server cluster), you can create a generic “deploy” project that manages the common deployment stage. Individual application projects can execute the deploy project, thereby eliminating a now-redundant build step.
**Distributed builds**

At sites that support dozens of applications, each with its own dependencies and build steps, it’s easy to inadvertently create dependency conflicts and bottlenecks because too many pipelines are running at once. To compensate, Jenkins lets you graduate to a distributed build architecture. This mode of operation uses a “build master,” a central system that keeps track of all the projects and their current state, and “build agents,” which run the actual build steps for a project. If you use Jenkins a lot, you’ll move to this configuration pretty quickly.

Build agents run on hosts that are separate from the build master. The Jenkins master logs in to the slaves (usually through SSH) to start the agent process and to add labels that document the slaves’ capabilities. For example, you might distinguish your Java-capable agents from your C-capable agents by applying appropriate labels.

For best results, run agents in containers, remote VMs, or ephemeral cloud instances that scale out and back on demand. If you have a container cluster, you can use Jenkins plug-ins to run agents in the cluster through a container management system.
Pipeline as code

Thus far, we’ve described the process of setting up Jenkins projects by stringing together individual build steps in the web UI. This is the quickest way to get started with Jenkins, but from an infrastructure perspective it’s also a bit opaque. The “code”—in this context, the contents of each build step—is managed by Jenkins. You can’t check graphical build steps into a code repository, and if you lose the Jenkins server, there’s no easy way to replace it; you’ll need to restore your projects from a recent backup.

Jenkins version 2 introduced a major new feature, called the Pipeline, that affords first-class support for CI/CD pipelines. A Jenkins pipeline codifies the steps of a project in a declarative, domain-specific language that’s based on the Groovy programming language. You can commit the Jenkins pipeline code, called a Jenkinsfile, alongside the code that’s associated with the pipeline.

The following Jenkinsfile demonstrates a basic build/test/deploy cycle:

```
pipeline {
    agent any
    stages {
        stage('Build') {
            steps {
                sh 'make'
            }
        }
        stage('Test') {
            steps {
                sh 'make test'
            }
        }
        stage('Deploy') {
            steps {
                sh 'deploy.sh'
            }
        }
    }
}
```

The agent any notation instructs Jenkins to prepare a workspace for this pipeline on any available build agent. A workspace is the same as a build context: a location on the agent’s local disk that contains all the files needed by the build, including the source code and dependencies. Every build has a private workspace.

The Build, Test, and Deploy stages parallel the conceptual stages of a CI/CD pipeline. In our example, each stage has a single step that invokes a shell (sh) to execute a command.
The Deploy stage runs a custom script, `deploy.sh`, that handles the entire deployment, including copying the build artifact (generated by the Build stage) to a set of servers and restarting server processes. In practice, deployment would usually be divided into multiple stages to afford better visibility and control over the full process.
26.4 CI/CD IN PRACTICE

We now turn to a contrived example to illustrate the concepts presented so far. We’ve concocted a simple application, UlsahGo, that’s a lot more basic than anything you might need to manage in the real world. It’s entirely self-contained and has no dependencies on other applications.

Our example includes the following elements:

- The UlsahGo web application, with just one small feature
- Unit tests for the application
- A virtual machine image for DigitalOcean, which contains the application
- A single-server development environment, created on demand
- A load-balanced, multiserver staging environment, created on demand
- A CI/CD pipeline that ties all these parts together

We use several popular tools and services in this example:

- GitHub as the code repository
- DigitalOcean virtual machines and load balancers
- HashiCorp’s Packer, for provisioning the DigitalOcean image
- HashiCorp’s Terraform, to create deployment environments
- Jenkins, to manage the CI/CD pipeline

Your applications might use a different technology stack, but the general concepts are similar, regardless of the tooling.

Exhibit C depicts the first several stages of the example pipeline. The diagram shows the pipeline polling GitHub for new commits to the UlsahGo project. When a commit is found, Jenkins runs the unit test suite. If the tests pass, Jenkins builds the binary. If the binary builds successfully, the pipeline continues to create the deployment artifact, a DigitalOcean machine image that includes the binary. If any of the stages fail, the pipeline reports an error.

Exhibit C: Demonstration pipeline (part one)
We describe the deployment stages in detail later. But first we should review these initial stages.
UlsahGo, a trivial web application

Our example application is a web service with a single feature. It returns, as JSON, the authors associated with a specified edition of this book. For example, the following query shows the authors for this edition:

```
$ curl ulsahgo.admin.com/?edition=5
{
    "authors": [
        "Evil",
        "Garth",
        "Trent",
        "Ben",
        "Dan"
    ],
    "number": 5
}
```

We do some sanity checking to make sure users aren’t getting too carried away, for example, by requesting implausible editions:

```
$ curl -vs ulsahgo.admin.com/?edition=6
HTTP/1.1 404 Not Found
< Content-Type: application/json
{
    "error": "6th edition is invalid."
}
```

Our application also has a health-check endpoint. Health checks are an easy way for monitoring systems to ask the application, “Hey, are you working OK?”

```
$ curl ulsahgo.admin.com/healthy
{
    "healthy": "true"
}
```

Developers typically work closely with administrators to create the build and test stages of a CI/CD pipeline. In this case, since the application is written in Go, we can use the standard Go tools (go build and go test) in our pipeline.
Unit testing UlsahGo

Unit tests are the first test suite to run because they operate at the source code level. Unit tests involve testing the application’s functionality at the finest possible granularity: its functions and methods. Most languages have testing frameworks that offer native support for unit tests.

Let’s examine one unit test for UlsahGo. Consider the following function:

```go
func ordinal(n int) string {
    suffix := "th"
    switch n {
        case 1:
            suffix = "st"
        case 2:
            suffix = "nd"
        case 3:
            suffix = "rd"
    }
    return strconv.Itoa(n) + suffix
}
```

The function takes an integer as input and determines the corresponding ordinal expression. For example, when passed a 1, the function returns “1st.” UlsahGo uses this function to format the text in the error message for invalid editions.

Unit tests try to prove that given some input, the function returns the expected output. Here’s a unit test that exercises this function:

```go
func TestOrdinal(t *testing.T) {
    ord := ordinal(1)
    exp := "1st"
    if ord != exp {
        t.Errorf("expected %s, got %s", exp, ord)
    }

    ord = ordinal(10)
    exp = "10th"
    if ord != exp {
        t.Errorf("expected %s, got %s", exp, ord)
    }
}
```

This simplified unit test runs the function on two values, 1 and 10, and confirms that the actual response matches the expectation. (The `ordinal()` function implements three special cases and a general case. A full set of unit tests would exercise each of these possible paths through the code.)

We can run the tests through Go’s built-in testing framework:
If some part of the application changes in the future—for example, if updates are made to the `ordinal()` function—the tests report any divergence from the expected output. Developers are responsible for updating unit tests as they adjust the code. Experienced developers by design write code that is easy to test. They aim to have complete coverage of each method and function.
Taking first steps with the Jenkins Pipeline

With the code ready to ship and the unit tests in place, the first step in our CI/CD journey is to configure the project in Jenkins. The GUI interface walks us through the process. Here are our selections:

- Our new project is a Pipeline project, defined by code as opposed to a traditional “freestyle” project with build steps mostly defined through user interface elements.

- We want to track our pipeline alongside our source code repository in a Jenkinsfile, so we choose “Pipeline script from SCM” for the Pipeline definition.

- We trigger the build by polling GitHub for commits. We add credentials so that Jenkins can access the UlsahGo repository and configure Jenkins to poll GitHub for changes every five minutes.

The initial setup takes just a few moments. In real life, we would use a GitHub web hook to notify Jenkins that a new commit was available, thus avoiding polling and sparing GitHub from our unnecessary calls to their API.

With this setup, Jenkins executes the pipeline described by the Jenkinsfile in the repository whenever a new commit is pushed to GitHub.

Consider how your repositories are organized. In this project we chose to combine CI/CD and application code in the same repository, with all CI/CD-related files being kept in the pipeline subdirectory. The UlsahGo repository is laid out as follows:

```bash
$ tree ulsahgo
ulsahgo
   pipeline
      Jenkinsfile
      packer
         provisioner.sh
         ulsahgo.json
         ulsahgo.service
      production
         tf_prod.sh
         ulsahgo.tf
      testing
         tf_testing.sh
         ulsahgo.tf
      ulsahgo.go
      ulsahgo_test.go
```

An integrated structure works well for a small project like this one. Jenkins, Packer, Terraform, and other tools can look in the pipeline subdirectory for their configuration files. Modifying the deployment pipeline is a simple matter of updating the repository. For more complex
environments where multiple projects share a common infrastructure, it makes sense to have a dedicated infrastructure repository.

With the project in place, we can commit our first Jenkinsfile. The first step in any pipeline is to check out the source code. Here’s a complete Jenkinsfile pipeline script that does just that:

```groovy
pipeline {
    agent any
    stages {
        stage('Checkout') {
            steps {
                checkout scm
            }
        }
    }
}
```

The checkout scm line instructs Jenkins to check out the code from “software configuration management,” a generic industry term for source control.

With Jenkins polling GitHub and the checkout stage complete, we can move on to setting up the test and build stages. Our Go project has no external dependencies. The only requirement for building and testing our code is the go binary. We have already installed go on our Jenkins system (with `apt-get -y install golang-go`) so we need only add the test and build stages to the Jenkinsfile:

```groovy
stage('Unit tests') {
    steps {
        sh 'go test'
    }
}
stage('Build') {
    steps {
        sh 'go build'
    }
}
```

After we commit the changes, Jenkins discovers the new commit and executes the pipeline. Jenkins emits friendly log output indicating that it has done so:

```
Mar 30, 2017 4:35:00 PM hudson.triggers.SCMTrigger$Runner run
INFO: SCM changes detected in UlsahGo. Triggering #4
    finish
INFO: UlsahGo #4 completed: SUCCESS
```

The Jenkins GUI uses a weather metaphor to indicate the health of recent builds. A sun icon represents a project that is building successfully, and a stormy cloud icon represents failures. You
can debug build failures by inspecting the console output, which is found under the build details. It shows the STDOUT printed by any part of the build.

Here’s a snippet from the go test and go build steps of our pipeline:

```
[Pipeline] stage
[Pipeline] { (Unit Tests)
[Pipeline]  sh
[Pipeline] Running shell script
  + go test
PASS
ok  _/var/jenkins_home/workspace/UlsahGo 0.006s
[Pipeline] }
[Pipeline] // stage
[Pipeline] stage
[Pipeline] { (Build)
[Pipeline]  sh
[Pipeline] Running shell script
  + go build
[Pipeline] }
[Pipeline] // stage
[Pipeline] }
[Pipeline] // node
[Pipeline] End of Pipeline
Finished: SUCCESS
```

You can normally pinpoint the cause of a failed build by reviewing the log. Look for error messages that identify the failed step. You can also add your own log messages to supply clues about the state of the system, such as the values of variables or the contents of a script at a given point in the execution. Writing output for the purpose of debugging is a time-honored programming tradition.

The output of our build is a single binary file, ulsahgo, which contains our entire application. (This, incidentally, is one of the primary benefits of Go programs and one of the reasons Go is popular with sysadmins: it’s easy to create static binaries that run on multiple architectures and have no external dependencies. Installing a Go application is often as simple as copying it to the system.)
Building a DigitalOcean image

With ulsahgo ready to ship, we next build a virtual machine image for the DigitalOcean cloud. We start with a vanilla Ubuntu 16.04 image, install the latest updates, and then install ulsahgo. The resulting image becomes the deployment artifact for the remaining stages of the pipeline.

If you’re unfamiliar with the tool packer, which creates virtual machine images, refer to the section Packer before continuing.

packer reads its image configuration from a template that has two primary sections: builders, which interact with remote APIs to create machines and images, and optional provisioners, which run custom configuration steps.

The template for our UlsahGo image has only one builder:

```
"builders": [{
    "type": "digitalocean",
    "api_token": "rj8FsrMI17vqTlB8qqBn9f7xQedJkkZJ7cqJcB105nm06ihz",
    "region": "sfo2",
    "size": "512mb",
    "image": "ubuntu-16-04-x64",
    "snapshot_name": "ulsahgo-latest",
    "ssh_username": "root"
}]
```

The builder tells packer which platform to build the image on and how to authenticate to the API, among other provider-specific details.

Three provisioning steps follow:

```
"provisioners": [
    {
      "type": "file",
      "source": "ulsahgo",
      "destination": "/tmp/ulsahgo"
    },
    {
      "type": "file",
      "source": "pipeline/packer/ulsahgo.service",
      "destination": "/etc/systemd/system/ulsahgo.service"
    },
    {
      "type": "shell",
      "script": "pipeline/packer/provisioner.sh"
    }
]
```

See this page for more information about systemd unit files.
The first two provisioning steps add files to the image. The first file is the application itself, `ulsahgo`, which is uploaded to `/tmp` for later use. The second is a `systemd` drop-in unit file that manages the service.

The last provisioner executes a custom shell script on the remote system. The script, `provisioner.sh`, updates the system and then sets up the application:

```bash
#!/usr/bin/env bash
app=ulsahgo

# Update the OS and add a user
apt-get update && apt-get -y upgrade
/usr/sbin/useradd -s /usr/sbin/nologin $app

# Set up the working directory and app
mkdir /opt/$app && chown $app /opt/$app
cp /tmp/$app /opt/$app/$app
chown $app /opt/$app/$app && chmod 700 /opt/$app/$app

# Enable the systemd unit
systemctl enable $app
```

In addition to shell scripts, `packer` lets you use all the popular configuration management tools as provisioning steps. Call out to Puppet, Chef, Ansible, or Salt to provision your images in a more structured and scalable manner.

Finally, we can add an image-building stage to our Jenkinsfile:

```groovy
stage('Build image') {
    steps {
        sh 'packer build pipeline/packer/ulsahgo.json > packer.txt'
        sh 'grep ID: packer.txt | grep -E -o \'^[0-9]*\' > do_image.txt'
    }
}
```

The first step invokes `packer` and saves the output to `packer.txt` in the build’s working directory. The tail end of that output includes an ID for the new image:

```shell
=> digitalocean: Gracefully shutting down droplet...
=> digitalocean: Creating snapshot: ulsahgo-latest
=> digitalocean: Waiting for snapshot to complete...
=> digitalocean: Destroying droplet...
=> digitalocean: Deleting temporary ssh key...
Build 'digitalocean' finished.
=> Builds finished. The artifacts of successful builds are:
--- digitalocean: A snapshot was created: (ID: 23838540)
```

The second step `grep`s the ID from `packer.txt` and saves it to a new file in the build context. Because the image is the deployment artifact, we will need to refer to its ID from later stages of the pipeline.
Provisioning a single system for testing

At this point we have a process for continuously running unit tests, building the application, and creating a virtual machine image as a build artifact. The remaining build stages focus on deploying the artifact and testing it in the wild. Exhibit D picks up where Exhibit C leaves off.

Exhibit D: Demonstration pipeline (part two)

We’ve chosen to use terraform, another gem from HashiCorp, to create and manage the UlsahGo infrastructure. terraform reads its configuration from “plans,” JSON-like configuration files that describe a desired infrastructure configuration. It then creates the cloud resources described in the plan by making an appropriate series of API calls. terraform supports dozens of cloud providers and a wide variety of services.

The following terraform configuration, ulsahgo.tf, requisitions a single DigitalOcean droplet running the image we created in the previous stage of the pipeline:

```terraform
variable "do_token" {}
variable "ssh_fingerprint" {}
variable "do_image" {}

provider "digitalocean" {
  token = "${var.do_token}"
}

resource "digitalocean_droplet" "ulsahgo-latest" {
  image = "${var.do_image}"
  name = "ulsahgo-latest"
  region = "sfo2"
  size = "512mb"
  ssh_keys = ["${var.ssh_fingerprint}"]
}
```
Most of this is self-explanatory: use DigitalOcean as the provider, and authenticate with the provided token. Create the droplet in the sfo2 region from the specified image ID.

In the Packer template on this page, we directly embedded parameters such as the API token in the builder configuration. One (big!) problem with that approach is that the API key is saved in the source code repository even though it’s supposedly secret. The key gives access to the cloud provider’s API and hence would be dangerous in the wrong hands. Keeping secrets in revision control is a security antipattern for reasons we describe in more detail here.

In this example, we instead read the parameters as variables. The three variables are

- The DigitalOcean API token
- The fingerprint of an SSH key that will be permitted to access the droplet
- The ID of the image to use for the new system, which we captured during the previous stage of the pipeline.

Jenkins can store secrets such as the API token in its “credential store,” an encrypted area that’s intended for exactly this kind of sensitive data. The pipeline can read values from the credential store and save them as environment variables. The values then become accessible throughout the pipeline without being saved in the version control system.

Here’s how we set this up in the Jenkinsfile:

```java
pipeline {
    environment {
        DO_TOKEN = credentials('do-token')
        SSH_FINGERPRINT = credentials('ssh-fingerprint')
    }
    ...
}
```

Recall that we saved the ID of the DigitalOcean machine image to a file within the build area, `do_image.txt`. We need that ID in our new pipeline stage, which creates the actual DigitalOcean droplet. The code for the new stage just runs a script from the project repository:

```bash
stage('Create droplet') {
    steps {
        sh 'bash pipeline/testing/tf_testing.sh'
    }
}
```

It’s easier and more maintainable to separate the code of complex scripts from the rest of the pipeline, as we’ve done here. `tf_testing.sh` contains the following lines:
This script copies the saved image ID to a temporary directory, pipeline/testing, then runs `terraform` from that directory. `terraform` looks for files in the current directory that have .tf extensions, so we don’t have to explicitly name the plan file. (It’s the same `ulsahgo.tf` file that we looked at on this page.)

A few explanations:

- The DO_TOKEN and SSH_FINGERPRINT environment variables are available to any shell commands in the pipeline. The environment clause shown above can appear either at the level of the overall pipeline or within a particular stage, depending on the scope you want.

- `$(<do_image.txt)` reads the contents of the saved DigitalOcean image ID from the text file saved in the previous stage.

- The final line of the `tf_testing.sh` script inspects the `terraform`-created droplet, obtains its IP address, and saves the address to a text file for use in the next stage. The `terraform.tfstate` file is `terraform`’s snapshot of the system state. It’s how `terraform` keeps track of resources.

Like `packer`, `terraform` sends useful output to the Jenkins console output page. Here are the relevant bits from the `terraform apply` command:
[Pipeline] { (Create droplet)
[Pipeline] sh
[UlsahGo] Running shell script
digitalocean_droplet.ulsahgo-latest: Creating...
   disk:       "" => "<computed>"
   image:      "" => "23888047"
   ipv4_address:    "" => "<computed>"
   ipv4_address_private:    "" => "<computed>"
   name:       "" => "ulsahgo-latest"
   region:     "" => "sfo2"
   resize_disk:  "" => "true"
   size:       "" => "512mb"
   ssh_keys.#:    "" => "1"
   ssh_keys.0:    "" => "****"
   status:      "" => "<computed>"
digitalocean_droplet.ulsahgo-latest: Still creating... (10s elapsed)
digitalocean_droplet.ulsahgo-latest: Still creating... (20s elapsed)
digitalocean_droplet.ulsahgo-latest: Still creating... (30s elapsed)
digitalocean_droplet.ulsahgo-latest: Creation complete (ID: 44486631)
Apply complete! Resources: 1 added, 0 changed, 0 destroyed.

When this stage completes, the droplet is up and running with UlsahGo.
Testing the droplet

We have some confidence that the code is functional because it passed the unit testing step. However, we also need to make sure that it runs successfully as part of a DigitalOcean droplet. Testing at this level is considered a form of integration test. We want the integration tests to run each time a new image is created, so we add a new stage to the **Jenkinsfile**:

```bash
stage('Test and destroy the droplet') {
    steps {
        sh '''#!/bin/bash
        curl -D -v $(<do_ip.txt):8000/?edition=5 | grep "HTTP/1.1 200"
        curl -D -v $(<do_ip.txt):8000/?edition=6 | grep "HTTP/1.1 404"
        terraform destroy -force
    '''
}
```

Sometimes a blunt, heavy object is the right tool for the job. This pair of **curl** commands queries **ulsahgo** on the remote droplet’s port 8000, where **ulsahgo** runs by default. We check that a query for the fifth edition returns an HTTP code 200 (success) and that a query for the sixth edition returns an HTTP 404 (failure). We know to expect these specific status codes only because of our familiarity with the application.

At the conclusion of the tests, we destroy the droplet because it’s no longer needed. The droplet is created, tested, and destroyed each time the pipeline runs.
Deploying UlsahGo to a pair of droplets and a load balancer

The final pipeline task is to deploy to our (mock) production environment, which consists of two DigitalOcean droplets and a load balancer. Once again, Terraform is up to the task.

We can reuse some of the configuration from the single-droplet terraform plan file. We still need the same variables and the droplet resource. This time, we add a second droplet resource:

```terraform
resource "digitalocean_droplet" "ulsahgo-b" {
  name = "ulsahgo-b"
  size = "512mb"
  image = "${var.do_image}"
  ssh_keys = ["${var.ssh_fingerprint}"],
  region = "sfo2"
}
```

We also add a load balancer resource:

```terraform
resource "digitalocean_loadbalancer" "public"
{
  name = "ulsahgo-lb"
  region = "sfo2"
  forwarding_rule {
    entry_port = 80
    entry_protocol = "http"
    target_port = 8000
    target_protocol = "http"
  }
  healthcheck {
    port = 8000
    protocol = "http"
    path = "/healthy"
  }
  droplet_ids = [
    "${digitalocean_droplet.ulsahgo-a.id}"
    "${digitalocean_droplet.ulsahgo-b.id}"
  ]
}
```

The load balancer listens on port 80 and forwards requests to each of the droplets on port 8000, where ulsahgo is listening. We tell the load balancer to use the /healthy endpoint to confirm that each copy of the service is running. The load balancer adds a droplet to the rotation if it receives a 200 status code when it queries this endpoint.

Now we can add the production configuration as a new stage in the pipeline:
The load balancer stage is more or less identical to the single instance stage. Even the external script is pretty much the same, so we omit its contents here. We could easily refactor these scripts so that a single version could handle both environments, but for now, we’ve kept the scripts separate.

We can add a testing stage as well, this time running against the load balancer’s IP address:

```bash
stage('Test load balancer') {
    steps {
        sh ''#/bin/bash -l
        curl -D -v -s $(<do_lb_ip.txt)/?edition=5 | grep "HTTP/1.1 200"
            curl -D -v -s $(<do_lb_ip.txt)/?edition=6 | grep "HTTP/1.1 404"
    }
}
```

The `curl` commands are similar to the previous set, but they target port 80, where the load balancer listens.
Concluding the demonstration pipeline

This demonstration CI/CD implementation captures several of the key elements of a real-world pipeline:

- The first two stages (unit testing and building) demonstrate continuous integration. Each time a developer commits code, Jenkins runs unit tests and tries to build the project.
- The third stage (creating a DigitalOcean image as a build artifact) is the beginning of continuous delivery. We can use the same image when deploying to each environment.
- Deployment to a single droplet is considered a “development” or “testing” environment.
- The final stage deploys *ulsahgo* to a high-availability, production-like environment, thereby closing the loop on a continuous deployment pipeline.
- If any stage in the pipeline fails, the subsequent stages are skipped. In that case, console output is available to help debug the problem.

This pipeline relies on open source tools throughout. All the deployment code is captured in just a few text files that are kept in the same repository as the application’s source code.

Astute readers will think of a host of improvements that could be made to these steps. To name just a few:

- A blue/green deployment to ensure no downtime in the production stage
- Status notifications to email or chat rooms for each stage
- Hooks to help monitoring systems note that a new deployment has occurred
- A better method of propagating data, such as the image ID, between stages

Continuous improvement is integral to CI/CD (and to system administration in general). Over time, a chain of incremental improvements results in a highly efficient and automated software delivery system.
26.5 **Containers and CI/CD**

Most software relies on outside dependencies such as third party libraries, a particular filesystem layout, the availability of certain environment variables, and other localizations. Conflicts among required dependencies often make it hard to run multiple applications on a single virtual machine.

To further complicate matters, building an application requires resources different from those running it. For example, the build process might require a compiler and a test suite, but these extras are not needed at run time.

See Chapter 25 for more information about containers.

Containers offer an elegant solution to these problems. From an operations viewpoint, the environment needs only the capability to run containers. You can activate any given container on any container-compatible system without further configuration effort because all dependencies and localizations for an app are housed within its container. Multiple containers can run on the same system simultaneously without conflict.

You can use containers to simplify your CI/CD environment in several ways:

- By running the CI/CD system itself within a container
- By building applications inside containers
- By using container images as build artifacts for deployment

The first point is rather obvious: you can run your CI/CD software (including both the master and any agents) in containers, thereby avoiding the overhead of having to dedicate systems to the CI/CD infrastructure.

The other two scenarios require a bit more explication. We look at them in more detail in the next sections.
Containers as a build environment

The exact environment needed to build an application is project-specific and sometimes quite complex. Rather than installing all the necessary tools, build software, and dependencies directly on your CI/CD agent systems, you can build your software within containers and leave the CI/CD agents in a clean and generic state. The build process then becomes portable and independent of the specific CI/CD agent.

Consider a typical application that depends on a PostgreSQL database and a Redis key/value store. To build and test the application in a traditional setting, you’d need separate servers for each component: the application itself, the Redis daemon, and PostgreSQL. In a pinch, you might run all these components on one system, but you probably wouldn’t use that same server to build and test another service that had different dependencies.

Instead, you can use short-lived containers for each component. One container can build and run the application. It can connect to separate containers (on the same host or a different host) for PostgreSQL and Redis. Once the build process is complete, the containers can be stopped and discarded. You can use the same CI/CD agent to build, with no risk of conflicts, applications that have other dependencies.

The container image used to build software should be distinct from the container image that runs it. The build image is normally larger than the run-time image because it includes extra components such as compilers and testing tools.

Most current CI/CD tools include native support for containers. Jenkins has a Docker plug-in that integrates nicely with the pipeline. Also check out Drone (try.drone.io), a CI/CD platform designed around containers.
### Container images as build artifacts

The product of a build can be a container image deployable through a container orchestration system. Containers are lightweight and highly portable. Moving container images among systems by way of an image registry is easy and fast. Any CI/CD tool can adopt the strategy of producing containers.

The basic workflow becomes:

1. Build your application inside a build-specific container.
2. Create a container image that includes the application and its dependencies.
3. Push the image to a registry.
4. Deploy that image to a container-ready execution environment.

It’s generally best to use a container management platform such as Docker Swarm, Mesos/Marathon, Kubernetes, or AWS EC2 Container Service to deploy images into production. Your pipeline’s deployment stage can call the appropriate APIs and let the platform handle the specifics. Exhibit E illustrates the procedure.

### Exhibit E: Container-based deployment process

![Container-based deployment process diagram]

We’ve found containers to be an excellent match for mature CI/CD pipelines. Their extremely fast cycle time makes it easy both to deploy new code and to revert to a previous version in the event of a problem. Both virtual machines and configuration management systems are an order of magnitude slower.
26.6 RECOMMENDED READING

BECK, KENT, ET AL. Manifesto for Agile Software Development. agilemanifesto.org


FOWLER, MARTIN. Continuous Integration. goo.gl/Y2lisI (martinfowler.com)


Computer security is in a sorry state. In contrast to the progress seen in virtually every other area of computing, security flaws have become increasingly dire and the consequences of inadequate security more severe. Computer security issues directly influence and threaten societies around the world.

If you’re tempted to skip over this chapter, permit us to pique your curiosity by reminding you of a few computer security events that have occurred since the last edition of this book:

- The sophisticated Stuxnet worm, discovered in 2010, attacked Iran’s nuclear program by damaging centrifuges at a uranium enrichment plant.

- In 2013, Edward Snowden exposed the massive NSA surveillance machine, revealing that some major Internet companies were complicit in allowing the government to spy on U.S. citizens.

- Around 2013, a new type of attack known as ransomware came to prominence. Attackers compromise a target system and encrypt its data, holding it hostage. Victims must pay a ransom for recovery. They often do.

- In 2015, the U.S. Office of Personnel Management was breached, compromising the sensitive and private details of more than 21 million U.S. citizens, many of whom had security clearances.
In 2016, Russian state-sponsored hackers allegedly mounted a campaign to influence the outcome of the U.S. presidential election.

In 2017, a ransomware attack of unprecedented scale took over Windows systems in more than 150 countries. The attack used an exploit developed by the NSA.

The stakes have never been higher. We think it will get worse before it gets better.

Part of the challenge is that security problems are not purely technical. You cannot solve them by buying a particular product or service from a third party. Achieving an acceptable level of security requires patience, vigilance, knowledge, and persistence—not just from you and other sysadmins, but from your entire user and management communities.

As a system administrator you bear a heavy burden. You must push an agenda that secures your organization’s systems and networks, ensures that they are vigilantly monitored, and properly educates your users and your staff. Familiarize yourself with current security technology and work with experts to identify and resolve vulnerabilities at your site. Security considerations should be part of every decision.

Strike a balance between security and usability. Remember that

\[
\text{Security} = \frac{1}{(1.072)(\text{Convenience})}
\]

The more security measures you introduce, the more constrained you and your users will be. Implement the security measures suggested in this chapter only after carefully considering the implications for your users.

Is UNIX secure? Of course not. UNIX and Linux are not secure, nor is any other operating system that communicates on a network. If you must have absolute, total, unbreachable security, then you need a measurable air gap between your computer and any other device. Sometimes even an air gap isn’t enough. In a 2014 paper, Genkin, Shamir, and Tromer described a technique to extract RSA encryption keys from laptops by analyzing the high-pitched frequencies they emit when decrypting a file.

Some people argue that you also need to enclose your computer in a special room that blocks electromagnetic radiation (look up “Faraday cage”). How fun is that?

This chapter examines the complex field of computer security: the sources of attacks, the basic ways to secure systems, the tools of the trade, and sources of additional information.
27.1 Elements of Security

The field of information security is quite broad, but it is often best described by the “CIA triad.” This acronym stands for

- Confidentiality
- Integrity
- Availability

Confidentiality concerns the privacy of data. Access to information should be limited to those who are authorized to have it. Authentication, access control, and encryption are a few of the subcomponents of confidentiality. If a hacker breaks into a system and steals a database of customer contact information, a compromise of confidentiality has occurred.

Integrity relates to the authenticity of information. Data integrity technology ensures that information is valid and has not been altered in unauthorized ways. It also addresses the trustworthiness of information sources. When a secure web site presents a signed TLS certificate, it is proving to the user not only that the information it is sending is encrypted but also that a trusted certificate authority (such as VeriSign or Equifax) has verified the identity of the source. Technologies such as PGP also offer some assurance of data integrity.

Availability expresses the idea that information must be accessible to authorized users when they need it. Otherwise, the data has no value. Outages not caused by intruders (e.g., those caused by administrative errors or power outages) also fall into the category of availability problems. Unfortunately, availability is often ignored until something goes wrong.

Consider the CIA principles as you design, implement, and maintain systems and networks. As the old security adage goes, “security is a process.”
27.2 How security is compromised

In this section we take a general look at how real-world security problems tend to occur. Most security lapses fit into one of the following categories.
Social engineering

Human users (and administrators) of a computer system are the weakest links in the chain of security. Even in today’s world of heightened security awareness, unsuspecting users with good intentions are easily convinced to give away sensitive information. No amount of technology can protect against the user element—you must ensure that your user community has a high awareness of security threats so that they can be part of the defense.

This problem manifests itself in many forms. Attackers cold-call their victims and pose as legitimately confused users in an attempt to get help accessing the system. Someone unintentionally posts sensitive information on a public forum while troubleshooting problems. Physical compromises occur when seemingly legitimate maintenance personnel show up to rewire the network closet.

The term “phishing” describes attempts to collect information from users or to coax them to into doing something foolish, such as installing malware. Phishing begins with deceptive emails, instant messages, text messages, or social media contacts. Targeted attacks (so called “spear phishing”) can be especially hard to defend against because the communication often includes victim-specific information that lends an appearance of authenticity.

Social engineering is a powerful hacking technique and is one of the most difficult threats to neutralize. Your site security policy should include training for new employees. Regular, organization-wide communications are an effective way to inform staff about social media threats, physical security, email phishing, multifactor authentication, and good password selection.

To gauge your organization’s resistance to social engineering, you might find it informative to attempt some social engineering attacks of your own. Be sure you have explicit permission to do this from your own managers, however. These exploits look very suspicious if they are performed without a clear mandate! They’re also a form of internal spying, so they have the potential to generate resentment if they’re not handled forthrightly.

Many organizations find it useful to communicate to users that administrators will never request their passwords. Tell users to report any such password requests to the IT department immediately.
Software vulnerabilities

Over the years, countless security-sapping bugs have been discovered in computer software. By exploiting subtle programming errors or context dependencies, hackers have been able to manipulate systems into a variety of compromising positions.

Buffer overflows are an example of a programming error with complex security implications. Developers often allocate a predetermined amount of temporary memory space, called a buffer, to store a particular piece of information. If the code isn’t careful about checking the size of the data against the size of the container that’s supposed to hold it, the memory adjacent to the allocated space is at risk of being overwritten. Crafty hackers can input carefully composed data that crashes the program or, in the worst case, executes arbitrary code.

Buffer overflows are a subcategory of a larger class of software security bugs known as input validation vulnerabilities. Nearly all programs accept some type of input from users (e.g., command-line arguments, parameters for an HTTP request). If the code processes such data without rigorously checking it for appropriate format and content, bad things can happen.

In some ways, open source operating systems have a leg up on security. The source code for Linux and FreeBSD is available to everyone, and thousands of people can (and do) scrutinize each line of code for possible security threats. This arrangement is widely believed to result in better security than that of closed operating systems, where a limited number of people have the opportunity to examine the code for weaknesses.

See Chapter 25 for more information about containers.

What can you as an administrator do to prevent this type of attack? It depends on the application, but one obvious approach is to reduce the privileges that your applications run with to minimize the impact of security bugs. A process running as an unprivileged user can do less damage than one that runs as root. For the paranoid, this approach can include a mandatory access control system such as SELinux. Containers with limited capabilities can also play a role here.

Over time, the open source community has developed a standard process for addressing software vulnerabilities. Initial reports should go directly to the software developers so that patches to address the issue can be developed and released before hackers formulate methods to exploit it. Later, details of the security issue are released publicly so that administrators become aware of it and so that the issue and the patches can receive public scrutiny. For this reason, keeping up with patches and security bulletins is an important part of most administrators’ job. Fortunately, modern operating systems strive to make software updates straightforward and easy to automate.
Distributed denial-of-service attacks (DDoS)

A DDoS attack aims to interrupt a service or adversely impact its performance, making the service unavailable to users. It’s usually achieved by flooding a site with network traffic, thereby consuming all the site’s available bandwidth or system resources. DDoS attacks can be financially motivated (in which case the attacker holds the site for ransom), or they can be either political or retaliatory.

To conduct an attack, attackers plant malicious code on unprotected devices outside the victim’s network. This code lets the attackers remotely command these intermediary systems, forming a “botnet.” In the most common DDoS scenario, the minions of the botnet are instructed to pelt the victim with network traffic.

In recent years, botnets have been assembled from Internet-connected devices such as IP cameras, printers, and even baby monitors. These devices have essentially no security, and the owners usually remain unaware that their devices have been compromised. Sophisticated command-and-control tools for managing botnets are available on the dark web for anyone to purchase. Some of them even include free customer service!

In the fall of 2016, the Mirai botnet targeted security researcher and blogger Brian Krebs, slamming his site with 620 Gb/s of traffic from tens of thousands of source IP addresses. Naturally, his hosting provider asked him to kindly move somewhere else. The Mirai botnet code has since been open sourced.

Most of the responsibility for preventing and mitigating DDoS attacks falls to the network management layer. Software and hardware are available to detect attacks and shut them down while keeping legitimate services on-line. Public cloud providers and some co-location facilities are equipped with this technology. However, the mitigations aren’t perfect and the threats are constantly shifting.
Insider abuse

Employees, contractors, and consultants are trusted agents of an organization and are granted special privileges. Sometimes these privileges are abused. Insiders can steal or reveal data, disrupt systems for financial gain, or create havoc for political reasons.

This type of attack is often the hardest of all to detect. Most security measures guard against external threats, so they aren’t effective against users who have been granted access. Insiders are typically not under suspicion in the first place; only the most rigorous organizations systematically monitor their own employees.

System administrators must never knowingly install back doors in the environment for their own use. Such facilities are too easily misinterpreted or exploited by others.
Network, system, or application configuration errors

Software can be configured securely or not-so-securely. Software is developed to be useful instead of annoying, hence not-so-securely is too often the default. Hackers frequently gain access by exploiting features that would be considered helpful and convenient in less treacherous circumstances: accounts without passwords, firewalls with overly relaxed rules, and unprotected databases, to name a few.

A typical example of a host configuration vulnerability is the standard practice of allowing Linux systems to boot without requiring a boot loader password. GRUB can be configured at installation to require a password, but administrators almost never activate this option. This omission leaves the system open to physical attack.

However, it’s also a perfect example of the need to balance security against usability. Requiring a password means that if the system were unintentionally rebooted (e.g., after a power outage), an administrator would have to be physically present to get the machine running again.

One of the most important steps in securing a system is simply making sure that you haven’t inadvertently put out a welcome mat for hackers. Problems in this category are the easiest to find and fix, although there are potentially a lot of them, and it’s not always obvious what to check for. The port and vulnerability scanning tools covered later in this chapter can help a motivated administrator identify problems before they’re exploited.
27.3 **Basic security measures**

Most systems do not come secured out of the box. Customizations made during and after installation change the security profile for new systems. Administrators should take steps to harden new systems, integrate them into the local environment, and plan for their long-term security maintenance.

When the auditors come knocking, it’s useful to be able to prove that you have followed some kind of standard procedure, especially if that procedure conforms to external recommendations and best practices for your industry. Refer to *Sources of security information* for recommendations on selecting a system-hardening standard.

At the highest level, you can improve your site’s security by keeping in mind a few rules of thumb:

- Apply the principle of least privilege by allocating only the minimum privileges needed by each entity, person, or role. This principle applies to firewall rules, user permissions, file permissions, and any other situation where access controls are used.

- Layer security measures to achieve defense in depth. For example, don’t rely solely on your external firewall for network protection. Otherwise, you are simply building a structure like a Tootsie Pop: a hard, crunchy outside and a soft, chewy center.

- Minimize the attack surface. The fewer interfaces, exposed systems, unnecessary services, and unused or underused systems, the lower the potential for vulnerabilities and security weaknesses.

Automation is a close ally in the security war. Use configuration management and scripting to create repeatably secure systems and applications. The more security steps you automate, the less room is available for human error.
Software updates

Keeping systems updated with the latest patches is an administrator’s highest-value security chore. Most systems are preconfigured to point at the vendor’s package repository, which makes applying patches as simple as running a few commands. Larger sites can use a local repository that mirrors that of the vendor, thus saving external bandwidth and speeding updates.

A reasonable approach to patching should include the following elements:

- A regular schedule for installing routine patches that is followed diligently. Consider the impact of patches on users when designing this schedule. Monthly updates are usually sufficient, but be prepared to apply critical patches on short notice.

- A change plan that documents the impact of each set of patches, outlines post-installation testing steps, and describes how to back out the changes in the event of problems. Communicate this change plan to all relevant parties.

- An understanding of which patches pertain to the environment. Administrators should subscribe to vendor-specific security mailing lists and blogs, as well as to generalized security discussion forums such as Bugtraq.

- An accurate inventory of applications and operating systems used in your environment. This census helps ensure complete coverage. Use reporting software to keep track of your installed base.

Software patches sometimes introduce novel security problems and weaknesses of their own. However, most exploits target older vulnerabilities that are widely known. Statistically speaking, you are much better off with systems that are regularly updated. Make sure it’s done methodically and consistently.
Unnecessary services

Stock systems come with lots of services running by default. Disable (and possibly remove) those that are unnecessary, especially if they are network daemons. One way to see which services are using the network is to use the `netstat` command. Here’s partial output from a FreeBSD system:

```
freebsd$ netstat -an | grep LISTEN
    tcp6  0   0   *.*22  *.* LISTEN
    tcp6  0   0   *.*2049*.* LISTEN
    tcp6  0   0   *.*989*.* LISTEN
    tcp6  0   0   *.*111*.* LISTEN
```

Linux is transitioning to the `ss` command for this purpose, but `netstat` still works there, too.

A variety of commands can help pinpoint the service that’s using a port. For example, you can use `lsof`:

```
freebsd$ sudo lsof -i:22
COMMAND   PID USER   FD   TYPE SIZE/OFF NODE NAME
sshd  701   root  3u   IPv6  0t0  TCP *:*ssh (LISTEN)
sshd  701   root  4u   IPv4  0t0  TCP *:*ssh (LISTEN)
sshd  815   root  3u   IPv4  0t0  TCP 10.0.2.15:ssh->10.0.2.2:54834 (ESTABLISHED)
sshd  817  vagrant 3u   IPv4  0t0  TCP 10.0.2.15:ssh->10.0.2.2:54834 (ESTABLISHED)
```

See Chapter 2 for more about starting processes at boot time.

Once you have the PIDs, you can then use `ps` to identify specific processes. If a service is unneeded, stop it and make sure that it won’t be restarted at boot time. You can also use the tools `fuser` or `netstat -p` if `lsof` is not available.

Limit your systems’ overall footprint. The fewer packages, the less vulnerable software. The industry as a whole is beginning to address this issue by reducing the number of packages included in a default installation. Some specialized distributions such as CoreOS take this to the extreme and force nearly everything to run in a container.
Remote event logging

See Chapter 10 for more information about logging.

The syslog service forwards log information to files, lists of users, or other hosts on your network. Consider setting up a secure host to act as a central logging machine that parses forwarded events and responds appropriately. A single centralized log aggregator can capture logs from a variety of devices and alert administrators whenever meaningful events occur. Remote logging also prevents hackers from covering their tracks by rewriting or erasing log files on systems that have been compromised.

Most systems come configured to use syslog by default, but you will need to customize the configuration to set up remote logging.
**Backups**

Regular, tested system backups are an essential part of any site security plan. They fall into the “availability” bucket of the CIA triad. Make sure that all filesystems are regularly replicated and that you store some backups off-site. If a significant security incident occurs, you’ll then have an uncontaminated checkpoint from which to restore.

However, backups can also be a security hazard. Protect your backups by limiting (and monitoring) access and by encrypting backup files.
**Viruses and worms**

UNIX and Linux have been mostly immune from viruses. Only a handful exist (most of which are academic in nature), and none have wreaked the kind of costly havoc that has become commonplace in the Windows world. Nonetheless, this fact hasn’t stopped certain antivirus vendors from predicting the demise of the platform from malware—unless you purchase their antivirus product at a special low price, of course.

The exact reason for the lack of malicious software is unclear. Some claim that UNIX simply has less market share than its desktop competitors and is therefore not an interesting target for virus authors. Others insist that UNIX’s access-controlled environment limits the damage from self-propagating worms and viruses.

The latter argument has some validity. Because UNIX restricts write access to system executables at the filesystem level, unprivileged user accounts cannot infect the rest of the environment. Unless virus code is being run by root, the scope of infection is significantly limited. The moral, then, is not to use the root account for day-to-day activities. See this page for more comments on this issue.

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*See Chapter 18 for more information about email content scanning.*

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Perhaps counterintuitively, one valid reason to run antivirus software on UNIX servers is to protect your site’s Windows systems from Windows-specific viruses. A mail server can scan incoming email attachments for viruses, and a file server can scan shared files for infection.

ClamAV by Tomasz Kojm is a popular, free antivirus product for UNIX and Linux. This widely used GPL tool is a complete antivirus tool kit that includes signatures for thousands of viruses. You can download the latest version from clamav.net.

Of course, one school of thought argues that antivirus software is itself counterproductive. Its detection and prevention rates are mediocre, and the cost of licensing and management are burdensome. All too frequently, antivirus software breaks other aspects of a system, resulting in a variety of tech support problems. Some compromises have even resulted from attacks on the antivirus infrastructure itself.

Recent versions of Microsoft Windows include a basic antivirus tool called Windows Defender. It’s not the quickest to detect new forms of malware, but it’s effective and relatively unlikely to interfere with other aspects of the system.
**Root kits**

The craftiest hackers try to cover their tracks and avoid detection. Often, they hope to continue using your system to distribute software illegally, probe other networks, or launch attacks against other systems. They often use “root kits” to help them remain undetected. Sony is notorious for having included root-kit-like capabilities in the copy protection software included on millions of music CDs.

Root kits are programs and patches that hide important system information such as process, disk, or network activity. They come in many flavors and vary in sophistication from simple application replacements (such as hacked versions of `ls` and `ps`) to kernel modules that are nearly impossible to detect.

Host-based intrusion detection software such as OSSEC is an effective way to monitor systems for the presence of root kits. File integrity monitoring tools, such as AIDE for Linux, can alert you to files that have changed unexpectedly. Root-kit-finding scripts (such as `chkrootkit`, chkrootkit.org) can identify known kits.

Although programs are available to help administrators remove root kits from a compromised system, the time it takes to perform a thorough cleaning would probably be better spent saving data and wiping the system. The most advanced root kits are aware of common removal programs and make attempts to subvert them.
Packet filtering

If you’re connecting a system to a network that has Internet access, you must install a packet-filtering router or firewall between the system and the outside world. The packet filter should pass only traffic for services that you specifically want to offer from that system. Limiting the public exposure of your systems is a first-line defense. Many systems do not need to be directly accessible to the public Internet.

In addition to firewalls at the Internet gateway, you can double up with host-based packet filters such as ipfw for FreeBSD and iptables (or ufw) on Linux. Determine which services run on the host, and open ports only for those services. In some cases, you can also limit which source addresses are allowed to connect to each port. Many systems need only one or two ports to be accessible.

If your systems are in the cloud, you can use security groups rather than physical firewalls. When designing security group rules, be as granular as possible. Consider adding outbound rules as well, to limit an attacker’s ability to make outbound connections from your hosts. See the platform-specific sections in Cloud networking for additional discussion of this topic.
Passwords and multifactor authentication

We’re simple people with simple rules. Here’s one: every account must have a password, and it needs to be something that can’t easily be guessed. Password complexity rules may be a hassle, but they exist for a reason. Guessable passwords are one of the leading sources of compromise.

One of our favorite trends from recent years is the proliferation of support for multifactor authentication (MFA) systems. These schemes validate your identity both through something you know (a password or passphrase) and something you have, such as a physical device, commonly a phone. Almost any interface can be protected with MFA, from UNIX shell accounts to bank accounts. Enabling MFA is an easy and powerful security win.

For a variety of reasons, MFA is now an absolute minimum requirement for any Internet-facing portal that gives access to administrative privileges. That includes VPNs, SSH access, and administrative interfaces to web applications. An argument can be made that single-factor (password-only) authentication is not acceptable for any user authentication, but you must secure at least all administrative interfaces with MFA. Fortunately, excellent cloud-based MFA services are available, such as Google Authenticator and Duo (duo.com).
**Vigilance**

To ensure the security of your system, regularly monitor its health, network connections, process table, and overall status (usually, daily). Do regular self-assessments with the power tools discussed later in this chapter. Security compromises tend to start with a small foothold and expand, so the earlier you identify an anomaly, the better off you’ll be. This is much easier said than done.

You might find it beneficial to work with an external firm to perform a comprehensive vulnerability analysis. These projects can draw your attention to issues that you hadn’t previously considered. At a minimum, they establish a baseline understanding of the areas in which you’re most exposed. Such engagements often reveal that hackers have already been nesting in the client’s network.
Application penetration testing

Applications that are exposed to the Internet need their own security precautions in addition to general system and network hygiene. Because of the widespread proliferation of vulnerability data and exploit tools, it’s a good idea to have all applications penetration tested to verify that they’ve been designed with security in mind and have appropriate controls in place.

Security is only as strong as the weakest link in the chain. If you have a secure network and system infrastructure, but an application running on that infrastructure allows access to sensitive data without a password (for example), you have won the battle but lost the war.

Penetration testing is a poorly defined discipline. Many companies that tout their penetration testing services focus mostly on smoke and mirrors. The Hollywood scenes of adolescent kids in windowless basements filled with 1980s-era terminals aren’t entirely inaccurate. Buyer beware.

Fortunately, the Open Web Application Security Project (OWASP) publishes information about common application vulnerabilities and methods for probing applications for these issues. Our recommendation is that you have a professional third party (who specializes in application penetration testing) perform a penetration test at launch and periodically throughout the life of an application. Make sure they adhere to the OWASP methodology.
27.4 Passwords and user accounts

See this page for more information about password cracking.

In addition to securing all Internet-facing privileged access through multifactor authentication, it’s important to select and manage passwords securely. In the world of sudo, administrators’ personal passwords are just as important as root passwords. More so, in fact: the more frequently a password is used, the more opportunities there are for it to be compromised through methods other than brute-force decryption.

From a narrowly technical perspective, the most secure password of a given length consists of a random sequence of letters, punctuation, and digits. Years of propaganda and picky web site password forms have convinced most people that this is the sort of password they ought to be using. But of course, they never do, unless they use a password vault to remember passwords on their behalf. Random passwords are simply impractical to commit to memory at the lengths needed to withstand brute-force attacks (12 characters or longer).

Because password security increases exponentially with length, your best bet is to use a very long password (a “passphrase”) that is unlikely to appear elsewhere but is easy to remember. You can throw in a misspelling or modified character for extra credit, but the general idea is to let the length do the heavy lifting for you.

For example, “six guests drank Evi’s poisoned wine” is an excellent passphrase. (Or at least, it was until it appeared in this book.) That’s true despite the fact that it consists mostly of common, lowercase, dictionary words, and despite the fact that the words are logically related and grammatically ordered.

The other core concept that all administrators and users must keep in mind is that a given passphrase should never be used for more than one purpose. It is all too common that a large breach occurs and usernames with passwords are exposed. If those usernames and passwords were used elsewhere, all those accounts are compromised now, too. Never use the same password across administrative boundaries (e.g., your personal banking site vs. social media).
Password changes

Change root and administrator passwords

- At least every six months
- Every time someone who had access to them leaves your site
- Whenever you wonder whether security might have been compromised

In the past, the conventional wisdom has been that passwords should be changed frequently to guard against the possibility of undetected compromises. However, password updates have their own risks, and they disrupt life for administrators. Competent hackers install backup access mechanisms as soon as they penetrate a site, so password changes are less helpful than they might initially seem.

It’s still advisable to make regularly scheduled changes, but don’t go overboard. If you really want to increase security, you’re better off obsessing about password quality.
Password vaults and password escrow

It’s often said that passwords “should never be written down,” but it’s perhaps more accurate to say that they should never be left accessible to the wrong people. As security maven Bruce Schneier has noted, a scrap of paper in an administrator’s wallet is relatively secure in comparison to most forms of Internet-connected storage.

A password vault is a piece of software (or a combination of software and hardware) that stores passwords for your organization in a more secure fashion than “Would you like Windows to remember this password for you?”

Several developments have made a password vault almost a necessity:

- The proliferation of passwords needed not just to log in to computers, but also to access web pages, configure routers and firewalls, and administer remote services
- The increasing need for strong passwords as computers get so fast that weak passwords are easily broken
- Regulations that require access to certain data to be traceable to a single person—no shared logins such as root

Password management systems became more popular in the wake of U.S. legislation that imposed additional requirements on sectors such as government, finance, and health care. In some cases, this legislation requires multifactor authentication.

Password vaults are also a great boon for sysadmin support companies who must securely and traceably manage passwords not only for their own machines but also for their customers’ machines.

Password vaults encrypt the passwords they store. Typically, every user has a separate vault password. (Just when you thought your password travails were over, now you have even more passwords to manage and worry about!)

Many password vault implementations are available. Free ones for individuals (e.g., KeePass) store passwords locally, give all-or-nothing access to the password database, and do no logging. Appliances suitable for huge enterprises (e.g., CyberArk) can cost tens of thousands of dollars. Many of the commercial offerings charge either by the user or by the number of passwords they remember.

The vault system we particularly like is 1Password from AgileBits (1password.com). 1Password comes from the mass-market world, so it includes polished, cross-platform UIs and integration with common web browsers. 1Password has a “teams” layer that extends this foundation of personal password management into the domain of organizational secrets.
Another system worth evaluating is Secret Server from Thycotic (thycotic.com). This system is browser-based and was designed from the ground up to serve the needs of organizations. It includes extensive management and auditing features along with role-based access control (see [this page](#)) and fine-grained permission options.

One useful feature to look for in a password management system is a “break the glass” option, so named in honor of the hotel fire alarm stations that tell you to break the glass and pull the big red lever in the event of an emergency. In this case, “breaking the glass” means obtaining a password that you wouldn’t normally have access to, with loud alarms being forwarded to other administrators. It’s a nice compromise between parsimonious password sharing (a normal best practice) and the realities of emergency fire fighting.

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See [this page](#) for more information about the **passwd** file.

---

Poor password management is a common security weakness. By default, the contents of the **/etc/passwd** and **/etc/shadow** files (or on FreeBSD, the **/etc/master.passwd** file) determine who can log in, so these files are the system’s first line of defense against intruders. They must be scrupulously maintained and free of errors, security hazards, and historical baggage.

UNIX allows users to choose their own passwords, and although this is a great convenience, it leads to many security problems. The comments in the section **Passwords and user accounts** apply equally to user passwords.

It is important to regularly verify (preferably daily) that every login has a password. Entries in the **/etc/shadow** file that describe pseudo-users such as “daemon” who own files but never log in should have a star or an exclamation point in their encrypted password fields. These do not match any password and thus prevent use of the account.

At sites that use a centralized authentication scheme such as LDAP or Active Directory, the same logic applies. Enforce password complexity requirements and lock out accounts after a few failed login attempts.
Password aging

Most systems that have shadow passwords also let you compel users to change their passwords periodically, a facility known as password aging. This feature may seem appealing at first glance, but it has several problems. Users often resent having to change their passwords, and since they don’t want to forget the new password, they choose something simple that is easy to type and remember. Many users switch between two passwords each time they are forced to change, or increment a digit in the password, defeating the purpose of password aging. PAM modules (see this page) can help enforce strong passwords to avoid this pitfall.

On Linux systems, the chage program controls password aging. Using chage, administrators can enforce minimum and maximum times between password changes, password expiration dates, the number of days to warn users before their passwords expire, the number of days of inactivity that are permissible before accounts are automatically locked, and more. The following command sets the minimum number of days between password changes to 2, sets the maximum number to 90, sets the expiration date to July 31, 2017, and warns the user for 14 days that the expiration date is approaching:

```
linux$ sudo chage -m 2 -M 90 -E 2017-07-31 -W 14 ben
```

Under FreeBSD, the pw command manages password aging parameters. This example sets the password validity period to 90 days and sets the expiration date to September 25, 2017.

```
freebsd$ sudo pw user mod trent -p 2017-09-25 -e 90
```
Group logins and shared logins

Any login that is used by more than one person is bad news. Group logins (e.g., “guest” or “demo”) are sure terrain for hackers to homestead and are prohibited in many contexts by federal regulations such as HIPAA. Don’t allow them at your site. However, technical controls can’t prevent users from sharing passwords, so education is the best enforcement tactic.
User shells

In theory, you can set the shell for a user account to be just about any program, including a custom script. In practice, the use of shells other than standards such as bash and tcsh is a dangerous practice. If you find yourself tempted to create such a login, you might consider a passphrase-less SSH key pair instead.
Rootly entries

The only distinguishing feature of the root login is its UID of zero. Since there can be more than one entry in the `/etc/passwd` file that uses this UID, there can be more than one way to log in as root.

A common way for a hacker to install a back door after having obtained a root shell is to edit new root logins into `/etc/passwd`. Programs such as `who` and `w` refer to the name stored in `utmp` rather than the UID that owns the login shell, so they cannot expose hackers that appear to be innocent users but are really logged in as UID 0.

Don’t allow root to log in remotely, even through the standard root account. Under OpenSSH, you can set the `PermitRootLogin` configuration option to `No` in the `/etc/ssh/sshd_config` file to enforce this restriction.

Because of `sudo` (see this page), it’s rare that you’ll ever need to log in as root, even on the system console.
27.5 Security Power Tools

Some of the time-consuming chores mentioned in the previous sections can be automated with freely available tools. Here are a few of the tools you’ll want to look at.
Nmap: network port scanner

Nmap’s main function is to check a set of target hosts to see which TCP and UDP ports have servers listening on them. (As described here, a port is a numbered communication channel. An IP address identifies an entire machine, and an IP address + port number identifies a specific server or network conversation on that machine.) Since most network services are associated with “well known” port numbers, this information tells you quite a lot about the software a machine is running.

Running Nmap is a great way to find out what a system looks like to someone on the outside who is trying to break in. For example, here’s a report from a production Ubuntu system:

```
ubuntu$ nmap -sT ubuntu.booklab.atrust.com
Starting Nmap 7.40 (http://insecure.org) at 2017-03-01 12:31 MST
Interesting ports on ubuntu.booklab.atrust.com (192.168.20.25):
Not shown: 1691 closed ports
  PORT       STATE      SERVICE
  25/tcp     open       smtp
  80/tcp     open       http
  111/tcp    open       rpcbind
  139/tcp    open       netbios-ssn
  445/tcp    open       microsoft-ds
  3306/tcp   open       mysql

Nmap finished: 1 IP address (1 host up) scanned in 0.186 seconds
```

By default, nmap includes the -sT argument to try to connect to each privileged or well-known TCP port on the target host in the normal way. Use the -p option to explicitly specify a range of ports to scan.

Once a connection has been established, nmap immediately disconnects, which is impolite but not harmful to a properly written network server.

From the example above, we can see that the host ubuntu is running two services that are likely to be unused and that have historically been associated with security problems: portmap (rpcbind) and an email server (smtp). An attacker would most likely probe those ports for more information as a next step in the information-gathering process.

The state column in nmap’s output shows open for ports that have servers listening, closed for ports with no server, unfiltered for ports in an unknown state, and filtered for ports that cannot be probed because of an intervening packet filter. nmap does not classify ports as unfiltered unless it is running an ACK scan. Here are results from a more secure server, secure.booklab.atrust.com:
In this case, it’s clear that the host is set up to allow SMTP (email) and an HTTP server. A firewall blocks access to other ports.

In addition to straightforward TCP and UDP probes, **nmap** also has a repertoire of sneaky ways to probe ports without initiating an actual connection. In most cases, **nmap** probes with packets that look like they come from the middle of a TCP conversation (rather than the beginning) and waits for diagnostic packets to be sent back. These stealth probes may be effective at getting past a firewall or at avoiding detection by a network security monitor on the lookout for port scanners. If your site uses a firewall (see [Firewalls](#)), it’s a good idea to probe it with these alternative scanning modes to see what they turn up.

**nmap** has the magical and useful ability to guess what operating system a remote system is running by looking at the particulars of its implementation of TCP/IP. It can sometimes even identify the software that’s running on an open port. The -O and -sV options, respectively, turn on this behavior. For example:

```bash
ubuntu$ sudo nmap -sV -O secure.booklab.atrust.com
```

Starting Nmap 7.40 (http://nmap.org) at 2017-03-01 12:44 MST
Interesting ports on secure.booklab.atrust.com (192.168.20.35):
Not shown: 1691 closed ports
PORT STATE SERVICE VERSION
25/tcp open smtp Postfix smtpd
80/tcp open http lighttpd 1.4.13
Device type: general purpose
Running: Linux 2.4.X|2.5.X|2.6.X
OS details: Linux 2.6.16 - 2.6.24

Nmap finished: 1 IP address (1 host up) scanned in 8.095 seconds

This feature can be useful for taking an inventory of a local network. Unfortunately, it is also useful to hackers, who can base their attacks on known weaknesses of the target OSes and servers.

Keep in mind that most administrators don’t appreciate your efforts to scan their network and point out its vulnerabilities, however well-intentioned your motive. Do not run **nmap** on someone else’s network without permission from one of that network’s administrators.
Nessus: next-generation network scanner

Nessus, originally released by Renaud Deraison in 1998, is a powerful and useful software vulnerability scanner. At this point, it uses more than 31,000 plug-ins to check for both local and remote security flaws. Although it is now a closed source, proprietary product, it is still freely available, and new plug-ins are released regularly. It is the most widely accepted and complete vulnerability scanner available.

Nessus prides itself on being the security scanner that takes nothing for granted. Instead of assuming that all web servers run on port 80, for instance, it scans for web servers running on any port and checks them for vulnerabilities. Instead of relying on the version numbers reported by the service it has connected to, Nessus can attempt to exploit known vulnerabilities to see if the service is susceptible.

Although a substantial amount of setup time is required to get Nessus running (it requires several packages that aren’t installed on a typical system), it’s well worth the effort. The Nessus system includes a client and a server. The server acts as a database and the client handles the GUI presentation. Nessus servers and clients exist for both Windows and UNIX platforms.

One of the great advantages of Nessus is the system’s modular design, which makes it easy for third parties to add new security checks. Thanks to an active user community, Nessus is likely to be a useful tool for years to come.
Metasploit: penetration testing software

Penetration testing is the act of breaking into a computer network with the owner’s permission for the purpose of discovering security weaknesses. Metasploit is an open source software package written in Ruby that automates this process.

Metasploit is controlled by the U.S.-based security company Rapid7, but its GitHub project has hundreds of contributors. Metasploit includes a database of hundreds of ready-made exploits for known software vulnerabilities. For those that have the desire and the skill, it’s possible to write custom exploit plug-ins to add to the database.

Metasploit uses the following basic workflow:

1. Scan remote systems to discover information about them.
2. Select and execute exploits according to the information found.
3. If a target is penetrated, use included tools to pivot from the compromised system to other hosts on the remote network.
4. Run reports to document the results.
5. Clean up and revert all changes to the remote system.

Metasploit has several interfaces: a command line, a web interface, and a full GUI client. Choose the format that you like the best; they have equivalent functionality. Learn more at metasploit.com.
Lynis: on-box security auditing

If you were faced with finding holes in the walls of an old wooden barn, you might first walk around the outside of the barn and look for the large, gaping holes. Network-based vulnerability scanning tools like Nessus give you this view of a system’s security profile. Walking inside the barn on a sunny day highlights the pinpoint-sized holes in walls. To get this same level of inspection of a system, you need a tool like Lynis that runs on the system itself.

Although unfortunately named, this security power tool performs both one-time and scheduled audits of a system’s configuration, patching, and hardening state. This open source tool runs on Linux and FreeBSD systems and performs hundreds of automated compliance checks. Download it from cisofy.com/lynis.
John the Ripper: finder of insecure passwords

One way to thwart poor password choices is to try to break the passwords yourself and to force users to change passwords that you have broken. John the Ripper is a sophisticated tool by Solar Designer that implements various password-cracking algorithms in a single tool. It replaces the tool crack, which was covered in previous editions of this book.

Even though most systems use a shadow password file to hide encrypted passwords from public view, it’s still wise to verify that your users’ passwords are crack resistant, especially the passwords of system administrators who have sudo privileges.

Knowing a user’s password can be useful because people tend to use the same password over and over again. A single password might grant access to another system, decrypt files stored in a user’s home directory, and allow access to financial accounts on the web. (Needless to say, it’s not security-smart to reuse a password this way. But nobody wants to remember hundreds of passwords.)

Considering its internal complexity, John the Ripper is an extremely simple program to use. Direct john to the file to be cracked, most often /etc/shadow, and watch the magic happen:

$ sudo ./john /etc/shadow
Loaded 25 password hashes with 25 different salts (FreeBSD MD5 [32/32])
password (jsmith)
badpass (tjones)

In this example, 25 unique passwords were read from the shadow file. As passwords are cracked, john prints them to the screen and saves them to a file called john.pot. The output contains the password in the left column with the login in parentheses in the right column. To reprint passwords after john has completed, run the same command with the -show argument.

As of this writing, the most recent stable version of John the Ripper is 1.8.0. It’s available from openwall.com/john. Since John the Ripper’s output contains the passwords it has broken, carefully protect this output and delete it as soon as you are done checking to see which users’ passwords are insecure.

As with most security monitoring techniques, it’s important to obtain explicit management approval before cracking passwords with John the Ripper.
**Bro: the programmable network intrusion detection system**

Bro is an open source network intrusion detection system (NIDS) that monitors network traffic and looks for suspicious activity. It was originally written by Vern Paxson and is available from bro.org.

Bro inspects all traffic flowing into and out of a network. It can operate in passive mode, in which it generates alerts for suspicious activity, or in active mode, in which it injects traffic to disrupt malicious activity. Both modes likely require modification of your site’s network configuration.

Unlike other NIDSs, Bro monitors traffic flows rather than just matching patterns inside individual packets. This method of operation means that Bro can detect suspicious activity by observing who talks to whom, even without matching any particular string or pattern. For example, Bro can

- Detect systems used as “stepping stones” by correlating inbound and outbound traffic
- Detect a server that has a back door installed by watching for unexpected outbound connections immediately after an inbound one
- Detect protocols running on nonstandard ports
- Report correctly guessed passwords

Some of these features require substantial system resources, but Bro includes clustering support to help you manage a group of sensor machines.

The configuration language for Bro is complex and requires significant coding experience to use. Unfortunately, there is no simple default configuration for a novice to install. Most sites require a moderate level of customization.

Bro is supported to some extent by the Networking Research Group of the International Computer Science Institute (ICSI), but it’s mostly maintained by the community of Bro users. If you are looking for a turnkey commercial NIDS, you will probably be disappointed by Bro. However, Bro can do things that no commercial NIDS can do, and it can either supplement or replace a commercial solution in your network.
Snort: the popular network intrusion detection system

Snort (snort.org) is an open source network intrusion prevention and detection system originally written by Marty Roesch and now maintained by Cisco, a commercial entity. It has become the de facto standard for home-grown NIDS deployments and is also the basis of many commercial and “managed services” NIDS implementations.

Snort itself is distributed for free as an open source package. However, Cisco charges a subscription fee for access to the most recent set of detection rules.

A number of third party platforms incorporate or extend Snort, and some of those projects are open source. One excellent example is Aanval (aanval.com), which aggregates data from multiple Snort sensors in a web console.

Snort captures raw packets off the network wire and compares them with a set of rules, aka signatures. When Snort detects an event that’s been defined as interesting, it can alert a system administrator or contact a network device to block the undesired traffic, among other actions.

Although Bro is a much more powerful system, Snort is a lot simpler and easier to configure, attributes that make it a good choice as a “starter” NIDS platform.
**OSSEC: host-based intrusion detection**

OSSEC is free software and is available as source code under the GNU General Public License. OSSEC serves up the following:

- Root kit detection
- Filesystem integrity checks
- Log file analysis
- Time-based alerting
- Active responses

OSSEC runs on the systems of interest and monitors their activity. It can send alerts or take action according to a set of rules that you configure. For example, OSSEC can monitor systems for the addition of unauthorized files and send email notifications like this one:

```
Subject: OSSEC Notification – courtesy – Alert level 7
Date: Fri, 03 Feb 2017 14:53:04 -0700
From: OSSEC HIDS <ossecm@courtesy.atrust.com>
To: <courtesy-admin@atrust.com>

OSSEC HIDS Notification.
2017 Feb 03 14:52:52

Received From: courtesy--syscheck
Rule: 554 fired (level 7) --> "File added to the system."
Portion of the log(s):

New file
    hbird.jpg'
added to the file system.

--END OF NOTIFICATION
```

In this way, OSSEC acts as your 24/7 eyes and ears on the system. We recommend running OSSEC on every production system.

**OSSEC basic concepts**

OSSEC has two primary components: the manager (server) and the agents (clients). You need one manager on your network, and you should install that component first. The manager stores the file-integrity-checking databases, logs, events, rules, decoders, major configuration options, and system auditing entries for the entire network. A manager can connect to any OSSEC agent, regardless of its operating system. The manager can also monitor certain devices that do not have a dedicated OSSEC agent.

Agents run on the systems you want to monitor and report back to the manager. By design, they have a small footprint and operate with a minimal set of privileges. Most of the agent’s
configuration is obtained from the manager. Communication between the server and the agent is encrypted and authenticated. You need to create an authentication key for each agent on the manager.

OSSEC classifies alerts by severity at levels 0 to 15; 15 is the highest severity.

**OSSEC installation**

OSSEC packages for most distributions are available at ossec.github.io.

Install the server on the system you want to be your OSSEC manager and then install the agent on that and all other systems you want to monitor. The install script asks some additional questions, such as to what email address alerts should be sent and which monitoring modules should be enabled.

Once the installation has finished, start OSSEC with

```
server$ sudo /var/ossec/bin/ossec-control start
```

Next, register each agent with the manager. On the server, run

```
server$ sudo /var/ossec/bin/manage_agents
```

You’ll see a menu that looks something like this:

```
********************************************
* OSSEC HIDS v2.8 Agent manager.
* The following options are available:
********************************************
   (A)dd an agent (A).
   (E)xtract key for an agent (E).
   (L)ist already added agents (L).
   (R)emove an agent (R).
   (Q)uit.
Choose your action: A,E,L,R or Q:
```

Select option A to add an agent, and then type in the name and IP address of the agent. Next, select option E to extract the agent’s key. Here’s what that looks like:

```
Available agents:
   ID: 001, Name: linuxclient1, IP: 192.168.74.3
Provide the ID of the agent to extract the key (or \d to quit): 001
Agent key information for '001' is:
MDAyIGxpbnV4Y2xpZz50MSAx0TIuMTY4LjTjMgZjk4YjMyYz1kMjg5MWJ1MT ...
```

Finally, log in to the agent system and run `manage_agents` there:
agent$ sudo /var/ossec/bin/manage_agents

On the client, you will see that the menu has somewhat different options.

******************************************
* OSSEC HIDS v2.8 Agent manager.
* The following options are available:
******************************************
   (I)mport key from the server (I).
   (Q)uit.
Choose your action: I or Q:

Select option I and then cut and paste the key you extracted above. After you have added an agent, you must restart the OSSEC server. Repeat the process of key generation, extraction, and installation for each agent you want to connect.

OSSEC configuration

Once OSSEC is installed and running, you’ll want to tweak it so that it gives you just enough information, but not too much. The majority of the configuration is stored on the server in the /var/ossec/etc/ossec.conf file. This XML-style file is well commented and fairly intuitive, but it contains dozens of options.

A common item you might want to configure is the list of files to ignore when checking file integrity. For example, if you have a custom application that writes its log file to /var/log/customapp.log, you can add the following line to the <syscheck> section of the file:

<syscheck>
   <ignore>/var/log/customapp.log</ignore>
</syscheck>

After you’ve made this change and restarted the OSSEC server, OSSEC will stop alerting you every time the log file changes. The many OSSEC configuration options are documented at ossec.net/main/manual/configuration-options.

It takes time and effort to get any HIDS system running and tuned. But after a few weeks, you’ll have filtered out the noise, and the system will start to generate valuable information about changing conditions in your environment.
Fail2Ban: brute-force attack response system

Fail2Ban is a Python script that monitors log files such as /var/log/auth.log and /var/log/apache2/error.log. It looks for suspicious occurrences such as multiple failed login attempts or queries to unusually long URLs. Fail2Ban then takes action to address the threat. For example, it might temporarily block network traffic from a particular IP address or send email to an incident response team. Learn more at fail2ban.org.
27.6 Cryptography Primer

Most software is designed with security in mind, and that implies a strong dose of cryptography. Security standards and regulations are opinionated about the selection of cryptographic algorithms and the type of data that must be protected with cryptography. Nearly all network protocols in modern use rely on cryptography for security. In short, cryptography is a pillar of computer security and sysadmins encounter it every day. It’s well worth your time to understand the basics.

Cryptography applies mathematics to the problem of securing communications. A cryptographic algorithm, called a cipher, is the set of mathematical steps taken to secure a message. Such algorithms are designed by committees of experts who represent academic, government, and research interests from around the world. Acceptance of a new algorithm is a lengthy and tedious process. By the time it makes its way to the masses, it has been thoroughly vetted.

Encryption is the process of using a cipher to convert plain text messages to unreadable ciphertext. Decryption is the reverse of that process. Cryptographic messages (ciphertext) exhibit several advantageous properties:

- **Confidentiality**: messages are impossible to read for everyone except the intended recipients.
- **Integrity**: it is impossible to modify the contents without detection.
- **Non-repudiation**: the authenticity of the message can be validated.

In other words, cryptography lets you communicate secretly over unsecured channels with the added benefit of being able to prove the correctness of the message and the identity of sender. Very valuable indeed.

Some ciphers offer only a subset of these features. Often, multiple ciphers are used together to complete the full set, thus forming a hybrid cryptosystem.

Mathematics shows that strong cryptographic algorithms are reliably secure. However, software that implements the algorithms might have weaknesses, and the security of systems that guard cryptographic secrets might also be vulnerable, rendering the algorithms impotent. Protecting your secrets and choosing well-designed, easily updated cryptography software is therefore paramount.

Cryptographers have traditional names for three subjects who participate in a simple message exchange: Alice and Bob, who wish to communicate privately, and Mallory, a bad actor who wants to compromise their secrets, disrupt their communication, or impersonate one of the other principals. We’ve adopted this convention.
The upcoming subsections introduce several cryptographic primitives, the associated ciphers, and some common use cases for each.
**Symmetric key cryptography**

Symmetric key cryptography is sometimes called “conventional” or “classic” cryptography because the ideas behind it have been around for a long time. It’s simple: Alice and Bob share a secret key that they use to encrypt and decrypt messages. They must find a way to exchange the shared secret privately. Once they both know the key, they can reuse it as long as they wish. Mallory can only inspect (or interfere with) messages if she also has the key.

Symmetric keys are relatively efficient in terms of CPU usage and the size of the encrypted payloads. As a result, symmetric cryptography is often used in applications where efficient encryption and decryption are necessary. However, the need to distribute the shared key in advance is a serious impediment to many use cases.

AES, the Advanced Encryption Standard from the United States National Institute of Standards and Technology (NIST), is perhaps the most widely used symmetric key algorithm. Twofish and its predecessor, Blowfish, designed by cryptographer and security expert Bruce Schneier, are also options. These algorithms play a role in the security of every network protocol you can shake your fist at, including SSH, TLS, IPsec VPNs, PGP, and many others.
Public key cryptography

A limitation of symmetric keys is the need to securely exchange the secret key in advance. The only way to do so with complete security is to meet in person without interference, a major inconvenience. For centuries, this requirement limited the practical utility of cryptography. The invention of public key cryptography, which addresses this problem, was therefore an extraordinary breakthrough when it occurred in the 1970s.

The scheme works as follows. Alice generates a pair of keys. The private key remains a secret, but the public key can be widely known. Bob similarly generates a key pair and publishes his public key. When Alice wants to send Bob a message, she encrypts it with Bob’s public key. Bob, who holds the private key, is the only one who can decrypt the message.

Exhibit A: Sending a ciphertext message with public key cryptography

<table>
<thead>
<tr>
<th>Alice</th>
<th>Encrypt with Bob’s public key</th>
<th>Ciphertext message</th>
<th>Decrypt with Bob’s private key</th>
<th>Plaintext message</th>
</tr>
</thead>
</table>

Alice can also sign the message with her private key. Bob can use Alice’s signature and her public key to validate its authenticity. This process (simplified here for clarity) is known as a digital signature. It proves that Alice, not Mallory, sent the message.

The Diffie-Hellman-Merkle key exchange method was the first publicly available public key cryptosystem. Shortly thereafter, the RSA public key cryptosystem was circulated by the now-famous team of Ron Rivest, Adi Shamir, and Leonard Adleman. These techniques are the foundation of modern network security.

Public key ciphers, also called asymmetric ciphers, rely on the mathematical concept of trapdoor functions, in which a value is easy to compute, and yet it is difficult and expensive to derive the steps that produced that value. The performance characteristics of asymmetric ciphers generally render them impractical for encrypting large quantities of data. They are often paired with symmetric ciphers to realize the benefits of both: public keys establish a session and share a symmetric key, and the symmetric key encrypts the ongoing conversation.
Public key infrastructure

Organizing a trustworthy and reliable way to record and distribute public keys is a messy business. If Alice wants to send Bob a private message, she must trust that the public key she has for Bob is in fact his and not Mallory’s. Validating the authenticity of public keys at Internet scale is a formidable challenge.

One solution, adopted by PGP, is a so-called web of trust. It boils down to a network of entities who trust each other to varying degrees. By following indirect chains of trust outside your personal network, you can establish that a public key is trustworthy with a reasonable degree of confidence. Unfortunately, the general public’s interest in attending key-signing parties and cultivating a network of cryptofriends has been, shall we say, less than enthusiastic, as evidenced by PGP’s continuing obscurity.

The Public Key Infrastructure, used to implement TLS on the web, addresses this problem by trusting a third party known as a Certificate Authority (CA) to vouch for public keys. Alice and Bob may not know each other, but they both trust the CA and know the CA’s public key. The CA signs certificates for Alice and Bob’s public keys with its own private key. Alice and Bob can then check the CA’s endorsements to be sure the keys are legitimate.

The certificates of major CAs such as GeoTrust and VeriSign are bundled with operating system distributions. When a client begins an encrypted session, it will see that the peer’s certificate has been signed by an authority already listed in the client’s local trust store. Hence the client can trust the CA’s signature and can trust that the peer’s public key is valid. The scheme is depicted in Exhibit B.

Exhibit B: Public key infrastructure process for the web

Certificate authorities charge a fee for signing services, the price of which is set according to the reputation of the CA, market conditions, and various features of the certificate. Some variations,

such as so-called wild card certificates for entire subdomains or “extended validation certificates” with a more rigorous background check for the requesting entity, are more expensive.

The CA is implicitly trusted in this system. Initially, there were only a few trusted CAs, but many more have been added over time. Modern desktop and mobile operating systems trust hundreds of certificate authorities by default. The CAs themselves are therefore high-value targets for attackers, who would like to use the CA’s private key to sign certificates of their own devising.

When an authority is hacked, the entire system of trust is broken. Several CAs are known to have been compromised by attackers, and in other widely discussed incidents, CAs are known to have conspired with governments. We encourage readers to choose issuing CAs carefully when purchasing signing services.

In 2016, Let’s Encrypt was launched as a free service (sponsored by organizations such as the Electronic Frontier Foundation, the Mozilla Foundation, Cisco Systems, Stanford Law School, and the Linux Foundation) that issues certificates through an automated system. By the end of 2016, this service had issued over 24 million certificates. Given the well-publicized operational issues at some of the commercial CAs, we recommend Let’s Encrypt as a “probably just as secure” free alternative.

It’s also easy to act as your own certificate authority. You can create a CA with OpenSSL, import the CA’s certificate to the trust store throughout your site, and then issue certificates against that authority. This is a common practice for securing services on an intranet where the organization has full control over the trusted certificate store. See this page for more details.

Organizations should be careful when deciding to implement their own trusted authority on company-issued machines. Unless you have the same rigorous and audited security in place that the professional CAs do, you might just be creating a gaping vulnerability in your environment. As a corollary, if you work for an organization that installs its own certificate in your computers’ trusted store, suspect that your own security may be compromised and act accordingly.
Transport Layer Security

Transport Layer Security (TLS) uses public key cryptography and PKI to secure messages between nodes on a network. It is the successor to SSL, the Secure Sockets Layer, and you’ll commonly see the acronyms SSL and TLS used interchangeably even though the old SSL is obsolete and deprecated. TLS paired with HTTP is known as HTTPS.

TLS runs as a separate layer that wraps TCP connections. It supplies only the security for the connection and does not involve itself in the HTTP transaction. Because of this hygienic architecture, TLS can secure not only HTTP but also other protocols such as SMTP.

Once a client and server have established a TLS connection, the contents of the exchange, including the URL and all headers, are protected by encryption. Only the host and port can be determined by an attacker since those details are visible through the encapsulating TCP connection. In the OSI model, TLS lies somewhere between layers 4 and 7.

Although the typical use case is one-way TLS encryption, in which the client validates the server, it is possible and increasingly common to use two-way TLS, sometimes known as mutual authentication. In this scheme, the client must present to the server a certificate that proves its own identity. This is, for example, how Netflix clients (set-top boxes and anything else that streams video from Netflix) are authenticated to the Netflix API.

The latest revision of TLS is 1.2. Disable all versions of SSL, along with TLS version 1.0, because of known weaknesses. TLS 1.3 is under active development and introduces major changes that will have significant implications for some industries.

A representative from the financial services industry attempted to influence a technical decision on the TLS development mailing list, but he was about two years too late. The concern was summarily rejected in an entertaining email thread. See the thread at goo.gl/uAEwPN.
Cryptographic hash functions

A hash function accepts input data of any length and generates a small, fixed-length value that is somehow derived from that data. The output value is variously referred to as a hash value, hash, summary, digest, checksum, or fingerprint. Hash functions are deterministic, so if you run a particular hash function on a particular input, you will always generate the same hash value.

Because hashes have a fixed length, only a finite number of possible output values exist. For example, an 8-bit hash has only $2^8$ (that is, 256) possible outputs. Therefore, some inputs necessarily generate the same hash value, an event known as a collision. Longer hash values reduce the frequency of collisions but can never eliminate them entirely.

Hundreds of different hash functions are used in software, but the subset known as cryptographic hash functions are of particular interest to sysadmins and mathematicians. In this context, “cryptographic” means “real good.” These hash functions are designed to have pretty much every desirable property you could want from a hash function, including the following:

- **Entanglement**: every bit of the hash value depends on every bit of the input data. On average, changing one bit of input should cause 50% of the hash bits to change.

- **Pseudo-randomness**: hash values should be indistinguishable from random data. Of course, hash values are *not* random; they are generated deterministically and reproducibly from input data. But they should still *look* like random data: they should have no detectable internal structure, should have no apparent relationship to the input data, and should pass all known statistical tests of randomness.

- **Nonreversibility**: given a hash value, it should be computationally infeasible to discover another input that generates the same hash value.

With a sufficiently high-quality hashing algorithm and a sufficiently long hash value length, we can make the leap of faith of assuming that two inputs that generate the same hash value are in fact the same input. Of course, that can’t ever be theoretically certain, because all hashes have collisions. However, it can be made likely to any desired level of statistical proof by increasing the length of the hash value.

Cryptographic hashes verify the integrity of things. They can certify that a given configuration file or command binary has not been tampered with, or that a message signed by an email correspondent has not been modified in transit. For example, to verify that a FreeBSD system and a Linux system are using identical `sshd_config` files, we can use the following commands:

```
freebsd$ sha256 /etc/ssh/sshd_config
SHA256 (/etc/ssh/sshd_config) = 3ef2d95099363d...8c14f63c5b9f741ea8d5

linux$ sha256sum /etc/ssh/sshd_config
3ef2d95099363d...8c14f63c5b9f741ea8d5 /etc/ssh/sshd_config
```
We’ve elided part of the hash values for simplicity. As is typical for most use cases, the output values are shown here in hexadecimal notation. But keep in mind that the actual hash values are just bags of binary data and that this data can be represented in multiple ways.

Many cryptographic hash algorithms exist, but the only ones recommended for general use at this point are the SHA-2 and SHA-3 (Secure Hash Algorithm) families, which were selected through an extensive review process by NIST. The older SHA-1 has been compromised and should no longer be used.

Each of these algorithms exists in a range of variants with different hash value lengths. For example, SHA3-512 is the SHA-3 algorithm configured to generate a 512-bit hash value. A SHA algorithm without a version number, e.g., SHA-256, always refers to a member of the SHA-2 family.

Another common cryptographic hash algorithm, MD5, remains widely supported by cryptographic software. However, it’s known to be vulnerable to engineered collisions, in which multiple inputs yield the same hash value. Although MD5 is no longer considered safe for use in cryptography, it’s still a well-behaved hash function and is theoretically OK to use for low-security applications. But why bother? Just use SHA.

Open source software projects often publish hashes of the files they release to the community. The OpenSSH project, for example, distributes PGP signatures (which rely on cryptographic hash functions) of its tarballs for verification. To verify the authenticity and integrity of a download, you calculate the hash value of the file you actually downloaded and compare it to the published hash value, thus ensuring that you’ve received a complete and unmolested copy with no bit errors.

Hash functions are also used as a component of message authentication codes, aka MACs. The hash value inside a MAC is signed with a private key. The process of validating the MAC checks both the authenticity of the MAC itself (by decrypting it with the corresponding public key) and the integrity of the content (by checking it against the content hash). MAC schemes often play an important role in web application security.
Random number generation

Cryptographic systems need a source of random numbers from which to generate keys. But algorithms aren’t known for their random and unpredictable behavior. What to do?

The gold standard for randomness is data from physically random processes such as radioactive decay and RF noise from the galactic core. These sources do exist: see random.org for access to some actual random data and an explanation of how it’s derived. Interesting, but unfortunately not directly helpful for day-to-day cryptography.

Traditional “pseudo-random” number generators use methods similar to those of hash functions to generate sequences of random-looking data. However, the process is deterministic. Once you know the internal state of the random number generator, you can reproduce the output sequence exactly. Ergo, this is usually a poor option for cryptography. When you generate a random 2048-bit key, you want 2048 bits’ worth of randomness, not 128 bits of number generator state that’s been algorithmically massaged into occupying 2048 bits.

Fortunately, kernel developers have put considerable effort into recording subtle variations in system behavior and using these as sources of randomness. Sources include everything from the timing of packets seen on a network to the timing of hardware interrupts to the vagaries of communication with hardware devices such as disk drives. Even on virtual and cloud servers, there’s still enough entropy available in the environment to generate reasonably random numbers.

All these sources feed forward into a secondary pseudo-random number generator that ensures the output stream of random data will have reasonable statistical properties. That data stream is then made available through a device driver. In Linux and FreeBSD, it’s presented as /dev/random and /dev/urandom.

Two main things to know about random numbers:

- Nothing that runs in user space can compete with the quality of the kernel’s random number generator. Never allow cryptographic software to generate its own random data; always make sure it uses random data from /dev/random or /dev/urandom. Most software does this by default.

- The choice of /dev/random vs. /dev/urandom is a matter of dispute, and unfortunately, the arguments are too subtle and mathematical to summarize here. The short version is that /dev/random on Linux is not guaranteed to generate data at all if the kernel feels that the system has not been accumulating enough entropy. Either get educated and pick one side or the other, or just use /dev/urandom and don’t worry your pretty little head about this issue. Most experts seem to recommend the latter approach. FreeBSD users are excused from battle, as /dev/random and /dev/urandom on the BSD kernel are identical.
Cryptographic software selection

There is good reason to be highly suspicious of all security software, and the packages that provide cryptographic services most of all. Major international governments are rumored to have attempted to influence the design phases of cryptographic protocols and algorithms. It seems safe to assume that several well-funded groups are eager to compromise any cryptographic project that is not fully nailed down.

That said, we trust open source software more than closed. Projects such as OpenSSL have a history of serious vulnerabilities, but those problems are disclosed, mitigated, and released in a transparent, open forum. The project history and source code are examined by thousands of people.

Never rely on home-grown cryptography of any sort. It is difficult enough just to use libraries correctly! Bespoke cryptosystems are doomed to vulnerability.
The openssl command

openssl is an administrator’s TLS multitool. You can use it to generate public/private key pairs, encrypt and decrypt files, examine the cryptographic properties of remote systems, create certificate authorities, convert among file formats, and myriad other cryptographic operations.

Preparing keys and certificates

One of the most common administrative functions of openssl is to prepare certificates for signing by a CA. Start by creating a 2048-bit private key:

$ openssl genrsa -out admin.com.key 2048

Use the private key to create a certificate signing request. openssl prompts for metadata known as the Distinguished Name (DN) to include with the request. It’s also possible to present this information in an answers file instead of in-line, as shown below.

$ openssl req -new -sha256 -key admin.com.key -out admin.com.csr
Country Name (2 letter code) [AU]:US
State or Province Name (full name) [Some-State]:Oregon
Locality Name (eg, city) []:Portland
Organization Name (eg, company) [Internet Widgits Pty Ltd]:ULSAH5E
Organizational Unit Name (eg, section) []:Crypto division
Common Name (e.g. server FQDN or YOUR name) []:server.admin.com

Submit the contents of admin.com.csr to the CA. The CA will perform a validation process to confirm that you are associated with the domain for which you’re obtaining a certificate (usually by sending email to an address within that domain), and will subsequently return a signed certificate. You can then use admin.com.key and the CA-signed certificate in your web server configuration.

Most of these fields are fairly arbitrary, but the Common Name is important. It must match the name of the subdomain you want to serve. If, for instance, you want to serve TLS for www.admin.com, make that your Common Name. You can request multiple names for a single certificate or a wild card that matches all the names in a subdomain; for example, *.admin.com.

Once you have the certificate, you can examine its properties. Here are some of the details of a wild card certificate for *.google.com:
The validity period is from Dec 15, 2016 through March 9, 2017. Clients who connect outside of this window will see error messages that the certificate is no longer valid. Tracking and managing certificate expiration dates is a common sysadmin responsibility.

Debugging TLS servers

Use `openssl s_client` to examine the TLS details of a remote server. This information can be quite useful when you are debugging web servers having certificate problems. For example, to examine the TLS properties of google.com (output truncated for brevity):

```
$ openssl s_client -connect google.com:443
---
New, TLSv1/SSLv3, Cipher is AES128-SHA
Server public key is 2048 bit
Secure Renegotiation IS supported
Compression: NONE
Expansion: NONE
SSL-Session:
    Protocol : TLSv1
    Cipher : AES128-SHA
    Session-ID: 4F72DC56EE4E80568F7E0EF9F59C8D7855C87F366B49BF1D9808...
    Session-ID-ctx:
    Master-Key: 895C6D08AF9B6B81E3E16BA65C0C9ACFACD72EF3335A32B86F3D3...
    Key-Arg : None
    Start Time: 1484163220
    Timeout : 300 (sec)
    Verify return code: 0 (ok)
---
```

You can use `openssl s_client` to check which versions of the TLS protocol a server supports. See also `openssl s_server`, which starts a generic TLS server. That can be handy for testing and debugging clients.
Phil Zimmermann’s PGP package provides a tool chest of bread-and-butter cryptographic utilities focused primarily on email security. It can encrypt data, generate signatures, and verify the origin of files and messages.

PGP has an interesting history that includes lawsuits, criminal prosecutions, and the privatization of portions of the original PGP suite. Recently, PGP has been heavily criticized for exposing too much metadata in its most common usage modes. Exposed message length, recipients, and cleartext draft storage (among other things) are weaknesses that could potentially be exploited by attackers, especially nation-state actors with generous resources. That said, PGP is still significantly better than sending information in plain text.

PGP’s file formats and protocols are being standardized by the IETF under the name OpenPGP, and multiple implementations of the proposed standard exist. The GNU project provides an excellent, free, and widely used implementation known as GnuPG at gnupg.org. For clarity, we refer to the systems collectively as PGP even though individual implementations have their own names.

Unfortunately, the UNIX and Linux versions are nuts-and-bolts enough that you have to understand a fair amount of cryptographic background to use them. Although you may find PGP useful in your own work, we don’t recommend that you support it for users because it has been known to spark many puzzled questions. We have found the Windows version to be considerably easier to use than the `gpg` command with its dozens of different operating modes.

Software packages on the Internet are often distributed with a PGP signature file that purports to guarantee the origin and purity of the software. However, it is difficult or impossible for people who are not die-hard PGP users to validate these signatures. Users must have collected a library of public keys from people whose identities they have personally verified. Downloading a single public key along with a signature file and software distribution is approximately as secure as downloading the distribution alone.

Some email clients add on a simple GUI for encrypted incoming and outgoing messages. Google Chrome users can install the “end to end” extension to incorporate PGP support for Gmail.
Kerberos: a unified approach to network security

The Kerberos system, designed at MIT, attempts to address some of the issues of network security in a consistent and extensible way. Kerberos is an authentication system, a facility that “guarantees” that users and services are in fact who they claim to be. It does not afford any additional security or encryption beyond that.

Kerberos uses symmetric and asymmetric cryptography to construct nested sets of credentials called “tickets.” Tickets are passed around the network to certify your identity and to give you access to network services. Each Kerberos site must maintain at least one physically secure machine (called the authentication server) on which to run the Kerberos daemon. This daemon issues tickets to users or services that present credentials (such as passwords) when they request authentication.

In essence, Kerberos improves on traditional password security in only two ways: it never transmits unencrypted passwords on the network, and it relieves users from having to type passwords repeatedly, making password protection of network services somewhat more palatable.

The Kerberos community boasts one of the most lucid and enjoyable documents ever written about a cryptosystem, Bill Bryant’s “Designing an Authentication System: a Dialogue in Four Scenes.” Despite its age it remains required reading for anyone interested in cryptography and is available at

web.mit.edu/kerberos/www/dialogue.html

Kerberos offers a better network security model than does the “ignoring network security entirely” model, but it is neither perfectly secure nor painless to install and run. It does not supersede the other security measures described in this chapter.

Unfortunately (and perhaps predictably), the Kerberos system distributed as part of Windows’ Active Directory uses proprietary, undocumented extensions to the protocols. As a result, it does not interoperate well with distributions based on the MIT code. Fortunately, the sssd daemon lets UNIX and Linux systems interact with Active Directory’s version of Kerberos. See the sections starting on this page for more information.
27.7 SSH, the Secure SHEll

The SSH system, invented by Tatu Ylönen, is a protocol for remote logins and for securing network services on an insecure network. SSH’s capabilities include remote command execution, shell access, file transfer, port forwarding, network proxy services, and even VPN tunneling. It is an indispensable tool, a veritable Swiss Army knife for system administrators.

SSH is a client/server protocol that uses cryptography for authentication, confidentiality, and integrity of communications between two hosts. It is designed for algorithmic flexibility, allowing the underlying cryptographic protocols to be updated and deprecated as the industry evolves. SSH is documented as a group of related protocols in RFCs 4250 through 4256.

In this section we discuss OpenSSH, the open source SSH implementation that is included and enabled by default on nearly every version of UNIX and Linux. We also mention a few alternative solutions for the adventurous and open-minded.
OpenSSH essentials

OpenSSH was developed by the OpenBSD project in 1999 and has since been maintained by that organization. The software suite consists of several commands:

- **ssh**, the client
- **sshd**, the server daemon
- **ssh-keygen**, for generating public/private key pairs
- **ssh-add** and **ssh-agent**, tools for managing authentication keys
- **ssh-keyscan**, for retrieving public keys from servers
- **sftp-server**, the server process for file transfer over SFTP
- **sftp** and **scp**, file transfer client utilities

In the most common and basic usage, a client establishes a connection to the server, authenticates itself, and subsequently opens a shell to execute commands. Authentication methods are negotiated according to mutual support and the preferences of the client and server. Many users can log in simultaneously. A pseudo-terminal is allocated for each, connecting their input and output to the remote system.

To initiate this process, a user invokes **ssh** with the remote host as the first argument:

```
$ ssh server.admin.com
```

**ssh** attempts a TCP connection on port 22, the standard SSH port assigned by IANA. When the connection is established, the server sends its public key for verification. If the server isn’t already known and trusted, **ssh** prompts the user to confirm the server by presenting a hash of the server’s public key called the key fingerprint:

```
The authenticity of host 'server.admin.com' can't be established.
ECDSA key fingerprint is SHA256:quLdFoXBI60pU6HwnUy/K50cR9UuU.
Are you sure you want to continue connecting (yes/no)?
```

The intent is that a server administrator can communicate the host key to users in advance. Users can then compare the information they received from the administrator to the server’s proffered fingerprint when they first connect. If the two match, the host’s identity is proved.

Once the user accepts the key, the fingerprint is added to `~/.ssh/known_hosts` for future use. **ssh** won’t mention the server’s key again unless the key changes, in which case **ssh** displays a nasty warning message that the server’s identity has changed.
In practice, this server verification dance is routinely ignored. Administrators rarely send the host key to users, and users blindly accept the host key without verification. This rubber-stamping of new host keys subjects users to man-in-the-middle attacks. Fortunately, the process can be automated and streamlined. We discuss this issue in *Host key verification with SSHFP*.

Once the host key has been accepted, the server lists the authentication methods it supports. OpenSSH implements all the methods described by the SSH RFCs, including simple UNIX password authentication, trusted hosts, public keys, GSSAPI for integration with Kerberos, and a flexible challenge/response scheme to support PAM and one-time passwords. Of these, public key authentication is the most commonly used and is the method we recommend for most sites. It offers the best balance of strong security and convenience. We discuss the use of public keys with SSH in more detail in *Public key authentication*.

**ssh** and **sshd** can be tuned for varying needs and security types. Configuration is found in the `/etc/ssh` directory, an uncharacteristically standard location among all flavors of UNIX and Linux. Table 27.1 enumerates the files found in that directory.

<table>
<thead>
<tr>
<th>File</th>
<th>Permissions</th>
<th>Contents</th>
</tr>
</thead>
<tbody>
<tr>
<td>ssh_config</td>
<td>0644</td>
<td>Site-wide client configuration</td>
</tr>
<tr>
<td>sshd_config</td>
<td>0644</td>
<td>Server configuration</td>
</tr>
<tr>
<td>moduli</td>
<td>0644</td>
<td>Prime numbers and generators for the DH key exchange</td>
</tr>
<tr>
<td>*_key</td>
<td>0600</td>
<td>Private keys for every algorithm supported by the server</td>
</tr>
<tr>
<td>*_key.pub</td>
<td>0644</td>
<td>A public key to match each private key</td>
</tr>
</tbody>
</table>

In addition to `/etc/ssh`, OpenSSH uses `~/.ssh` for storing public and private keys, for per-user client configuration overrides, and for a few other purposes. The `~/.ssh` directory is ignored unless its permissions are set to 0700.

OpenSSH has a respectable though not impeccable track record for security vulnerabilities. According to the CVE database (cve.mitre.org), several critical vulnerabilities were discovered in early versions. The last of these was documented in 2006. Occasional denial-of-service and bypass vulnerabilities continue to be announced, but most of them are considered relatively low risk. Still, it’s wise to update the OpenSSH packages as part of your regular patching schedule.
The ssh client

Getting started with ssh is straightforward, but its power and versatility lie in its many options. Through configuration you can choose cryptographic algorithms and ciphers, create convenient host aliases, set up port forwarding, and much more.

The basic syntax is

```
ssh [options] [username@]host [command]
```

For example, to check the disk space of /var/log:

```
$ ssh server.admin.com "df -h /var/log"
```

If you specify a command, ssh authenticates itself to the host, runs the command, and exits without opening an interactive shell. If you do not specify a username, ssh uses your local username on the remote host.

ssh reads configuration settings from the site-wide file /etc/ssh/ssh_config and processes additional options and overrides on a per-user basis from ~/.ssh/config. Table 27.2 lists some of the more interesting options that you can tune in these files. We discuss some of these options in more detail later in this chapter.

**Table 27.2: Useful SSH client configuration options**

<table>
<thead>
<tr>
<th>Option</th>
<th>Meaning</th>
<th>Default</th>
</tr>
</thead>
<tbody>
<tr>
<td>AddKeysToAgent</td>
<td>Automatically add keys to ssh-agent</td>
<td>no</td>
</tr>
<tr>
<td>ConnectTimeout</td>
<td>Connection timeout in seconds</td>
<td>varies *</td>
</tr>
<tr>
<td>ControlMaster</td>
<td>Allow connection multiplexing</td>
<td>no</td>
</tr>
<tr>
<td>DynamicForward</td>
<td>Set up a SOCKS4 or SOCKS5 proxy</td>
<td>–</td>
</tr>
<tr>
<td>ForwardAgent</td>
<td>Enable ssh-agent forwarding</td>
<td>no</td>
</tr>
<tr>
<td>Host</td>
<td>Marker for a new host alias</td>
<td>–</td>
</tr>
<tr>
<td>IdentityFile</td>
<td>Path to an authentication private key</td>
<td>~/.ssh/id_rsa b</td>
</tr>
<tr>
<td>Port</td>
<td>Port to connect on</td>
<td>22</td>
</tr>
<tr>
<td>RequestTTY</td>
<td>Specify whether a TTY is needed</td>
<td>auto</td>
</tr>
<tr>
<td>ServerAliveInterval</td>
<td>Pings for connections to the server</td>
<td>0 (disabled)</td>
</tr>
<tr>
<td>StrictHostKeyChecking</td>
<td>Require (yes) or ignore (no) host keys</td>
<td>ask</td>
</tr>
</tbody>
</table>

* The default is determined by the kernel's TCP defaults, which vary widely.

b. The precise name depends on the authentication algorithm. By default, all keys begin with id_.

When ssh assembles a final configuration, command-line arguments take precedence over entries in ~/.ssh/config. The global configuration set in /etc/ssh/ssh_config is the lowest-priority source of configuration options.
ssh sends the current username as the login name if another value is not specified. You can supply a different username with the -l flag or the @ syntax:

```
$ ssh -l hsolo server.admin.com
$ ssh hsolo@server.admin.com
```

Client options that are not available as direct arguments to ssh can still be set on the command line with the -o flag. For example, you could disable host checks for a server:

```
$ ssh -o StrictHostKeyChecking=no server.admin.com
```

The -v option prints debug messages. Specify it multiple times (maximum of three) to increase verbosity. You’ll find this flag to be invaluable when debugging authentication problems.

For convenience, ssh returns the exit status of the remote command. Use this behavior to check for error conditions when invoking ssh from scripts.

Consult **man ssh** and **man ssh_config** to familiarize yourself with available options and features. Run **ssh -h** for a succinct summary.
Public key authentication

OpenSSH (and the SSH protocol generally) can use public key cryptography to authenticate users to remote systems. As a user, you start by creating a public/private key pair. You give the public key to the server administrator, who adds it to the server in the file ~/.ssh/authorized_keys. You can then log in to the remote server by running ssh with the remote username and matching private key.

$ ssh -i ~/.ssh/id_ecdsa hsolo@server.admin.dom

Use ssh-keygen to generate a key pair. You can specify which cryptographic algorithm to use, as well as bit length and other characteristics. For example, to generate an ECDSA key pair with a 384-bit elliptic curve size:

$ ssh-keygen -t ecdsa -b 384
Generating public/private ecdsa key pair.
Enter file in which to save the key (/home/ben/.ssh/id_ecdsa): <return>
Enter passphrase (empty for no passphrase): <return>
Enter same passphrase again: <return>
Your identification has been saved in /home/ben/.ssh/id_ecdsa.
Your public key has been saved in /home/ben/.ssh/id_ecdsa.pub.
The key fingerprint is:
SHA256:VRh6raUfpm3YdtMm7GURbIoypc/pnbwhsmvscsr1hk4 ben

The public key (~/.ssh/id_ecdsa.pub) and private key (~/.ssh/id_ecdsa) files are base64-encoded ASCII files. Never share the private key! ssh-keygen sets the permissions on the public and private key correctly as 0644 and 0600, respectively. This example uses ECDSA, but it’s also fine to use -t rsa with 2048 or 4096 bits.

ssh-keygen prompts for an optional passphrase to encrypt the private key. If you use a passphrase, you must type it to decrypt the key before ssh can read it. A passphrase improves security because the authentication process gains an additional verification step: you must both have the key file and know the passphrase that decrypts it before you can authenticate.

We suggest setting a passphrase on all privileged accounts (that is, those with sudo privileges). If you need to use a key without a passphrase to enable an automated process, limit the corresponding server account’s permissions.

If you’re the server administrator and you need to add a public key for a new user, follow these steps:

1. Ensure that the user has an active account with a valid shell.
2. Get a copy of the user’s public key from the user.
3. Create the user’s .ssh directory with permissions 0700.
4. Add the public key to `~user/.ssh/authorized_keys` and set the permissions of that file to 0600.

For example, if user hsolo’s public key were saved in `/tmp/hsolo.pub`, the process would look like this:

```
$ grep hsolo /etc/passwd
hsolo:x:503:503:Han Solo:/home/hsolo:/bin/bash
$ mkdir -p ~hsolo/.ssh && chmod 0700 ~hsolo/.ssh
$ cat /tmp/hsolo.pub >> ~hsolo/.ssh/authorized_keys
$ chmod 0600 ~hsolo/.ssh/authorized_keys
```

If you do this more than once you’ll almost certainly find it prudent to script the procedure. Configuration management systems like Ansible and Chef handle this task cleanly.
The ssh-agent

The **ssh-agent** daemon caches decrypted private keys. You load your private keys into the agent, and **ssh** then automatically offers those keys when it connects to new servers, simplifying the process of connecting.

Use the **ssh-add** command to load a new key. If the key requires a passphrase, you’ll be prompted to enter it. To list the currently loaded keys, type **ssh-agent -l**:

```bash
$ ssh-add ~/.ssh/id_ecdsa
Enter passphrase for ~/.ssh/id_ecdsa: <passphrase>
Identity added: ~/.ssh/id_ecdsa (~/.ssh/id_ecdsa)
$ ssh-add -l
384 SHA256:VRbIoYfcp/npbwhsmvsdr1hK4 ~/.ssh/id_ecdsa (ECDSA)
```

You can have many keys active at once. Remove a key with **ssh-add -d path**, or purge all loaded keys with **ssh-add -D**.

Oddly, to remove the private key from the agent, the public key must be in the same directory and have the same filename but with a `.pub` extension. If the public key is not available, you might receive a confusing error message that the key does not exist:

```bash
$ ssh-add -d ~/.ssh/id_ecdsa
Bad key file /home/ben/.ssh/id_ecdsa: No such file or directory
```

You can easily fix this problem by extracting the public key with **ssh-keygen** and saving it to the expected filename. (This extraction is possible because the private key file contains a copy of the public key as well as the private key.)

```bash
$ key=/home/ben/.ssh/id_ecdsa
$ ssh-keygen -yf $key > $key.pub
Enter passphrase: <passphrase>
```

**ssh-agent** is even more useful when you leverage its key forwarding feature, which makes the loaded keys available to remote hosts while you are logged in to them through **ssh**. You can use this feature to jump from one server to another without copying your private key to remote systems. See **Exhibit C**.

**Exhibit C: ssh-agent forwarding**
To enable agent forwarding, either add `ForwardAgent yes` to your `~/.ssh.config` file or use `ssh -A`.

Use key forwarding only on servers that you trust. Anyone in control of the server you’ve forwarded to can assume your identity and access remote systems. They cannot read your private keys directly, but they can use any that are available through the forwarding agent.

<table>
<thead>
<tr>
<th>Client</th>
<th>Jump server</th>
<th>Private servers</th>
</tr>
</thead>
<tbody>
<tr>
<td>Runs <strong>ssh-agent</strong></td>
<td>aka bastion host</td>
<td>Connect from here to private servers without copying the key</td>
</tr>
</tbody>
</table>
**Host aliases in ~/.ssh/config**

You’ll undoubtedly encounter many different SSH configurations if you interact with or administer a large number of servers. To simplify your life, the ~/.ssh/config file lets you set up aliases and overrides for individual hosts.

For example, consider two systems. The first is a web server with IP address 54.84.253.153 where sshd listens on port 2222. Your username on that server is han and you have a private key for authentication. The other is debian.admin.com, where your username is hsolo. You’d prefer to disable password authentication entirely, but the Debian server requires it.

To connect to these servers from the command line you could use option-larded commands such as these:

```bash
$ ssh -l han -p 2222 -i /home/han/.ssh/id_ecdsa 54.84.253.153
$ ssh -l hsolo debian.admin.com
```

Your client in this case must leave password authentication enabled (the default) because it’s a hassle to type `-o PasswordAuthentication=no` all the time.

The following ~/.ssh/config sets up aliases for these hosts and has the added benefit of disabling password authentication by default:

```plaintext
PasswordAuthentication no
Host web
  HostName 54.84.253.153
  User han
  IdentityFile /home/han/.ssh/id_ecdsa
  ForwardAgent yes
  Port 2222
Host debian
  Hostname debian.admin.com
  User hsolo
  PasswordAuthentication yes
```

Now you can use the much simpler commands `ssh web` and `ssh debian` to reach these hosts. The client reads the aliases and sets options automatically for each system.

`ssh` also understands some basic patterns for matching hosts. For example:

```plaintext
Host *
  ServerAliveInterval 30m
  ServerAliveCountMax 1
Host 172.20.*
  User luke
```
This example tells ssh to keep idle connections open for 30 minutes on all servers. It also sets username “luke” when connecting to hosts on the 172.20/16 network.

Host aliases become more powerful than you can possibly imagine when combined with other tricks of the OpenSSH trade.
**Connection multiplexing**

`ControlMaster` is a nifty ssh feature that enables connection multiplexing, thus considerably improving SSH performance over WAN links. When enabled, the first connection to a host creates a socket that can be reused. Subsequent connections share the socket but require separate authentication.

Turn on multiplexing with the `ControlMaster`, `ControlPath`, and `ControlPersist` options in a Host alias:

```bash
Host web
  HostName 54.84.253.153
  User han
  Port 2222
  ControlMaster auto
  ControlPath ~/.ssh/cm_socket/%r%@%h:%p
  ControlPersist 30m
```

`ControlMaster auto` enables the feature. `ControlPath` creates a socket at the designated location. See `man ssh_config` for the substitutions that can be used in the `ControlPath` filename. In this case, the file is named according to the remote login username, host IP address, and port. Connecting to this host results in a socket like this one:

```
$ ls -l ~/.ssh/cm_socket/
  srw------- 1 ben  ben  0 Jan  2 15:22 han@54.84.253.153:22
```

Such a pattern guarantees a unique filename for each socket. `ControlPersist` saves the socket for the specified period of time even if the first connection (the “master”) disconnects.

Spend the 30 seconds it takes to set this up, then make a donation to the OpenBSD foundation to thank them for implementing multiplexing and saving you time.
**Port forwarding**

Another useful ancillary feature of SSH is its ability to tunnel TCP connections securely through an encrypted channel, thereby allowing connectivity to insecure or firewalled services at remote sites. Exhibit D shows a typical use of an SSH tunnel and should help clarify how it works.

**Exhibit D: An SSH tunnel for HTTP**

In this scenario, a remote user—let’s call her Alice—wants to establish an HTTP connection to a web server on an enterprise network. Access to that host or to port 80 is blocked by the firewall, but having SSH access, Alice can route the connection through the SSH server.

To set this up, Alice logs in to the remote SSH server with `ssh`. On the `ssh` command line, she specifies an arbitrary (but specific; in this case, 8000) local port that `ssh` should forward through the secure tunnel to the remote web server’s port 80.

```
$ ssh -L 8000:webserver:80 server.admin.com
```

All source ports in this example are marked as random since programs choose an arbitrary port from which to initiate connections.

To access the web server, Alice can now connect to port 8000 on her own machine. The local `ssh` receives the connection and tunnels Alice’s traffic over the existing SSH connection to the remote `sshd`. In turn, `sshd` forwards the connection to the web server on port 80.

Of course, tunnels such as these can be intentional or unintentional back doors as well. Use tunnels with caution and also watch for unauthorized use of this facility by users. You can disable port forwarding in `sshd` with the `AllowTCPForwarding no` configuration option.
sshd: the OpenSSH server

The OpenSSH server daemon, sshd, listens on port 22 (by default) for connections from clients. Its configuration file, `/etc/ssh/sshd_config`, boasts myriad options, some of which may need to be tuned for your site.

sshd runs as root. It forks an unprivileged child process for each connected client with the same permissions as the connecting user. If you make changes to the `sshd_config` file, you can force sshd to reload by sending a HUP signal to the parent process.

```
$ sudo kill -HUP $(sudo cat /var/run/sshd.pid)
```

In Linux, you can also run `sudo systemctl reload sshd`. The changes take effect for new connections. Existing connections are preserved without interruption but continue to use their previous settings.

The following example `sshd_config` includes some commonly adjusted options configured to balance server security with users’ convenience:

```bash
# Set to inet for IPv4-only or inet6 for IPv6-only
AddressFamily any

# Allows only the named users and groups to log in
# Somewhat draconian. Adding/removing users requires reload
AllowUsers foo bar hso10
AllowGroups admins

# TCP forwarding is convenient but can be abused
AllowTcpForwarding yes

# Display a message before users authenticate
# Important for inane legal reasons and compliance requirements
Banner /etc/banner

# We prefer to allow public key authentication only
ChallengeResponseAuthentication no
PasswordAuthentication no
RSAAuthentication no
GSSAPIAuthentication no
HostbasedAuthentication no
PubkeyAuthentication yes

# Disconnect inactive clients after 5 minutes
ClientAliveInterval 300
ClientAliveCountMax 1
```
We encourage you to list the acceptable ciphers and key exchange algorithms explicitly. We
don’t include the details here because the names are quite long and are a moving target anyway.
Follow Mozilla’s OpenSSH configuration guidelines, which can be found at goo.gl/Xxgx7H
(deep link into wiki.mozilla.org).

# Allow compression at all times
Compression yes

# Do not allow remote hosts to use forwarded ports
GatewayPorts no

# Record failed login attempts
LogLevel VERBOSE

# Reduced from the default of 6
MaxAuthTries 3

# Do not allow root to log in (encourages use of sudo)
PermitRootLogin no

# Prevent users from setting their environment in an authorized_keys file
PermitUserEnvironment no

# Use the "auth" facility for syslog messages
SyslogFacility AUTH

# Kill the session if a TCP connection is lost
TCPKeepAlive no

# Do not allow X forwarding if your site does not use X
X11Forwarding no
Host key verification with SSHFP

Recall from earlier in this section that SSH server host keys are routinely ignored by server administrators and users alike. Cloud instances exacerbate the problem because even the administrator has no knowledge of the host key before logging in.

Fortunately, a DNS record known as SSHFP has been developed to address this issue. The premise is that the server’s key is stored as a DNS record. When a client connects to an unknown system, SSH looks up the SSHFP record to verify the server’s key rather than asking the user to verify it.

The sshfp utility, available from github.com/xelerance/sshfp, generates SSHFP DNS resource records either by scanning a remote server (the -s flag) or by parsing a previously accepted key from a known_hosts file (the -k flag; this is also the default). Of course, either choice assumes that the source of the key is known to be correct.

For example, the following command generates a BIND-compatible SSHFP record for server.admin.com:

```
$ sshfp server.admin.com
server.admin.com IN SSHFP 1 1 94a26278ee713a37f6a78110f1ad9bd...
server.admin.com IN SSHFP 2 1 7cf72d02e3d3fa947712bc56fd0e0a3i...
```

Add these records to the domain’s zone file (be mindful of the names and the $ORIGIN), reload the domain, and use dig to verify the key:

```
$ dig server.admin.com. IN SSHFP | grep SSHFP
; <<< DiG 9.5.1-P2 <<< server.admin.com. IN SSHFP
; server.admin.com. IN SSHFP
server.admin.com. 38400 IN SSHFP 1 1 94a26278ee713a37f6a78110f...
server.admin.com. 38400 IN SSHFP 2 1 7cf72d02e3d3fa947712bc56f...
```

ssh does not consult SSHFP records by default. Add the VerifyHostKeyDNS option to /etc/ssh/ssh_config to enable checking. As with most SSH client options, you can also pass -o VerifyHostKeyDNS=yes on the ssh command line when you first access a new system.

You can automate this process by generating the SSHFP record in the server’s initialization scripts. Use dynamic DNS or your favorite DNS provider’s API to create the record.
File transfers

OpenSSH has two utilities for transferring files: `scp` and `sftp`. On the server side, `sshd` runs a separate process called `sftp-server` to handle file transfers. SFTP has no relationship to the older and insecure File Transfer Protocol, FTP.

You can use `scp` to copy files from your system to a remote host, from a remote host to your system, or between remote hosts. The syntax mirrors that of `cp` with some extra decorations to designate hosts and usernames.

$ scp ./file server.admin.com:
$ scp server.admin.com:file ./file
$ scp server1.admin.com:file server2.admincom:file

`sftp` is an interactive experience similar to a traditional FTP client. You can also find graphical SFTP interfaces for most desktop operating systems.
Alternatives for secure logins

Most systems and sites rely on OpenSSH for secure remote access, but it is not the only choice.

Dropbear is an SSH implementation with a focus on maintaining a compact footprint. It compiles to a statically linked 110KiB binary, perfect for consumer-grade routers and other embedded devices. It includes some of the same features as OpenSSH, such as public key authentication and agent forwarding.

Gravitational’s Teleport is another alternative SSH server that offers several advantages. Its authentication model relies on expiring certificates, which eliminates the problem of distributing and configuring users’ public keys. Among Teleport’s impressive features are an optional audit trail for each connection and a nifty collaboration system that lets multiple users share a session. Compared to OpenSSH, Teleport is relatively new and unproven, but to date there have been no reported vulnerabilities. We expect Gravitational to continue their rapid pace of development.

Mosh, developed by a brilliant team at MIT, is a replacement for SSH. Unlike SSH, Mosh operates on encrypted and authenticated UDP datagrams. It is designed for better performance over WAN connections and for roaming. For example, you can resume connections if you move from one IP address to another or if your connection drops. First released in 2012, Mosh has a much shorter history than OpenSSH, but in its first few years it has had no reported security vulnerabilities. Like Dropbear, it has a much smaller footprint than OpenSSH.
27.8 **Firewalls**

In addition to protecting individual machines, you can also implement security precautions at the network level. The basic tool of network security is the firewall, a device or piece of software that prevents unwanted packets from accessing networks and systems. Firewalls are ubiquitous today and are found in devices ranging from desktop systems and servers to consumer routers and enterprise-grade network appliances.
Packet-filtering firewalls

A packet-filtering firewall limits the types of traffic that can pass through your Internet gateway (or through an internal gateway that separates domains within your organization) according to information in the packet header. It’s much like driving your car through a customs checkpoint at an international border crossing. You specify which destination addresses, port numbers, and protocol types are acceptable, and the gateway simply discards (and in some cases, logs) packets that don’t meet the profile.

Packet-filtering software is included in Linux systems in the form of `iptables` (and its easier-to-use front end, `ufw`) and on FreeBSD as `ipfw`. See the details beginning [here](#) for more information.

Although these tools are capable of sophisticated filtering and bring a welcome extra dose of security, we generally discourage the use of UNIX and Linux systems as network routers and, most especially, as enterprise firewall routers. The complexity of general-purpose operating systems makes them inherently less secure and less reliable than task-specific devices. Dedicated firewall appliances such as those made by Check Point and Cisco are a better option for site-wide network protection.
Filtering of services

Most well-known services are associated with a network port in the /etc/services file or its vendor-specific equivalent. The daemons responsible for these services bind to the appropriate ports and wait for connections from remote sites. Most of the well-known service ports are “privileged,” meaning that their port numbers are in the range 1 to 1023. These ports can be used only by a process running as root (or with an appropriate Linux capability). Port numbers 1024 and higher are referred to as nonprivileged ports.

Service-specific filtering is predicated on the assumption that the client (the machine that initiates a TCP or UDP conversation) uses a non-privileged port to contact a privileged port on the server. For example, if you wanted to allow only inbound HTTP connections to a machine with the address 192.108.21.200, you would install a filter that allowed TCP packets destined for port 80 at that address and that permitted outbound TCP packets from that address to anywhere. (Port 80 is the standard HTTP port as defined in /etc/services.) The exact way that such a filter is installed depends on the kind of router or filtering system you are using.

Modern security-conscious sites use a two-stage filtering scheme. One filter is a gateway to the Internet, and a second filter lies between the outer gateway and the rest of the local network. The idea is to terminate all inbound Internet connections on systems that lie between these two filters. If these systems are administratively separate from the rest of the network, they can handle a variety of services for the Internet with reduced risk. The partially secured network is usually called the demilitarized zone or DMZ.

The most secure way to use a packet filter is to start with a configuration that allows no inbound connections. You can then liberalize the filter bit by bit as you discover useful things that don’t work and, hopefully, move any Internet-accessible services onto systems in the DMZ.
**Stateful inspection firewalls**

The theory behind stateful inspection firewalls is that if you could carefully listen to and understand all the conversations (in all languages) that were taking place in a crowded airport, you could make sure that someone wasn’t planning to bomb a plane later that day. Stateful inspection firewalls are designed to inspect the traffic that flows through them and compare the actual network activity to what “should” be happening.

For example, if the packets exchanged in an H.323 video sequence name a port to be used later for a data connection, the firewall should expect a data connection to occur only on that port. Attempts by the remote site to connect to other ports are presumably bogus and should be dropped.

So what are vendors really selling when they claim to deliver stateful inspection? Their products either monitor a limited number of connections or protocols or they search for a particular set of “bad” situations. Not that there’s anything wrong with that; clearly, some benefit is derived from any technology that can detect traffic anomalies. In this particular case, however, remember that the claims are *mostly* marketing hype.
**Firewalls: safe?**

A firewall should not be your only means of defense against intruders. It’s only one component of what ought to be a carefully considered, multilayered security strategy. Firewalls often confer a false sense of security. If a firewall lulls you into relaxing other safeguards, it will have had a *negative* effect on the security of your site.

Every host within your organization should be individually patched, hardened, and monitored with tools such as Bro, Snort, Nmap, Nessus, and OSSEC. Likewise, your entire user community needs to be educated about basic security hygiene.

Ideally, local users should be able to connect to any Internet service they like, but machines on the Internet should only be able to connect to a limited set of local services hosted within your DMZ. For example, you might want to allow SFTP access to a local archive server and allow SMTP connections to a server that receives incoming email.

To maximize the value of your Internet connection, we recommend that you emphasize convenience and accessibility when deciding how to set up your network. At the end of the day, it’s the system administrator’s vigilance that makes a network secure, not a fancy piece of firewall hardware.
27.9 Virtual private networks (VPNs)

In its simplest form, a VPN is a connection that makes a remote network appear as if it were directly connected, even if it is physically thousands of miles and many router hops away. For increased security, the connection is not only authenticated in some way (usually with a “shared secret” such as a passphrase), but the end-to-end traffic is also encrypted. Such an arrangement is usually referred to as a “secure tunnel.”

Here’s a good example of the kind of situation in which a VPN is handy. Suppose that a company has offices in Chicago, Boulder, and Miami. If each office has a connection to a local ISP, the company can use VPNs to transparently (and, for the most part, securely) connect the offices across the untrusted Internet. The company could achieve a similar result by leasing dedicated lines to connect the three offices, but that would be considerably more expensive.

Another good example is a company whose employees telecommute from their homes. VPNs let those users reap the benefits of their high-speed and inexpensive cable modem service while making it appear that they are directly connected to the corporate network.

Because of the convenience and popularity of this functionality, everybody is offering some type of VPN solution. You can buy it from your router vendor as a plug-in for your operating system or even as a dedicated VPN device for your network. Depending on your budget and scalability needs, you may want to consider one of the many commercial VPN solutions.

If you’re without a budget and looking for a quick fix, SSH can do secure tunneling for you. See Port forwarding.
IPsec tunnels

If you’re a fan of IETF standards (or of saving money) and need a real VPN solution, take a look at IPsec (Internet Protocol security). IPsec was originally developed for IPv6, but it has also been widely implemented for IPv4. IPsec is an IETF-approved, end-to-end authentication and encryption system. Almost all serious VPN vendors ship a product that has at least an IPsec compatibility mode. Linux and FreeBSD include native kernel support for IPsec.

IPsec uses strong cryptography to implement both authentication and encryption services. Authentication ensures that packets are from the right sender and have not been altered in transit, and encryption prevents the unauthorized examination of packet contents.

In tunnel mode, IPsec encrypts the transport layer header, which includes the source and destination port numbers. Unfortunately, this scheme conflicts with most firewalls. For this reason, most modern implementations default to transport mode, in which only the payloads of packets (the data being transported) are encrypted.

There’s a gotcha involving IPsec tunnels and MTU size. You must ensure that once a packet has been encrypted by IPsec, nothing fragments it along the network path the tunnel traverses. To achieve this feat, you might have to lower the MTU on the devices in front of the tunnel. (In the real world, 1,400 bytes usually works.) See this page in the TCP chapter for more information about MTU size.
All I need is a VPN, right?

Sadly, there’s a downside to VPNs. Although they do build a (mostly) secure tunnel across the untrusted network between the two endpoints, they don’t usually address the security of the endpoints themselves. For example, if you set up a VPN between your corporate backbone and your CEO’s home, you may inadvertently be creating a path for your CEO’s 15-year-old daughter to have direct access to everything on your network.

Bottom line: you need to treat connections from VPN tunnels as external connections and grant them additional privileges only as necessary and only after careful consideration. Think about adding a special section to your site security policy to cover the rules applying to VPN connections.
27.10 Certifications and Standards

If the subject matter of this chapter seems daunting to you, don’t fret. Computer security is a complicated and vast topic, as countless books, web sites, and magazines can attest. Fortunately, much has been done to help quantify and organize the available information. Dozens of standards and certifications exist, and mindful system administrators should reflect on their guidance.
Certifications

Large corporations often employ many full-time employees whose job is guarding information. To gain credibility in the field and keep their knowledge current, these professionals attend training courses and obtain certifications. Prepare yourself for acronym-fu as we work through a few of the most popular certifications.

One of the most widely recognized security certifications is the CISSP, or Certified Information Systems Security Professional. It is administered by (ISC)², the International Information Systems Security Certification Consortium (say that ten times fast!). One of the primary draws of the CISSP is (ISC)²’s notion of a “common body of knowledge” (CBK), essentially an industry-wide best practices guide for information security. The CBK covers law, cryptography, authentication, physical security, and much more. It’s an incredible reference for security folks.

One criticism of the CISSP has been its concentration on breadth and consequent lack of depth. So many topics in the CBK, and so little time! To address this, (ISC)² has issued CISSP concentration programs that focus on architecture, engineering, and management. These specialized certifications add depth to the more general CISSP certification.

The SANS Institute created the Global Information Assurance Certification (GIAC) suite of certifications in 1999. Three dozen separate exams cover the realm of information security with tests divided into five categories. The certifications range in difficulty from the moderate two-exam GISF to the 23-hour, expert-level GSE. The GSE is notorious as one of the most difficult certifications in the industry. Many of the exams focus on technical specifics and require quite a bit of experience.

Finally, the Certified Information Systems Auditor (CISA) credential is an audit and process certification. It focuses on business continuity, procedures, monitoring, and other management content. Some consider the CISA an intermediate certification that is appropriate for an organization’s security officer role. One of its most attractive aspects is that it involves only a single exam.

Although certifications are a personal endeavor, their application to business is undeniable. More and more companies now recognize certifications as the mark of an expert. Many businesses offer higher pay and promotions to certified employees. If you decide to pursue a certification, work closely with your organization to have it help pay for the associated costs.
Security standards

Because of the ever-increasing reliance on data systems, laws and regulations have been created to govern the management of sensitive, business-critical information. Major pieces of U.S. legislation such as HIPAA, FISMA, NERC CIP, and the Sarbanes-Oxley Act (SOX) have all included sections on IT security. Although the requirements are sometimes expensive to implement, they have helped give the appropriate level of focus to a once-ignored aspect of technology.

For a broader discussion of industry and legal standards that affect IT environments, see this page.

Unfortunately, the regulations are filled with legalese and can be difficult to interpret. Most do not contain specifics on how to achieve their requirements. As a result, standards have been developed to help administrators reach the lofty legislative requirements. These standards are not regulation-specific, but following them usually ensures compliance. It can be intimidating to confront the requirements of all the various standards at once, so it’s usually best to first work through one standard in its entirety.

ISO 27001:2013

The ISO/IEC 27001 (formerly ISO 17799) standard is probably the most widely accepted in the world. First introduced in 1995 as a British standard, it is 34 pages long and is divided into 11 sections that run the gamut from policy through physical security to access control. Objectives within each section define specific requirements, and controls under each objective describe the suggested best practice solutions. The document costs about $200.

The requirements are nontechnical and can be fulfilled by any organization in a way that best fits its needs. On the downside, the general wording of the standard leaves the reader with a sense of broad flexibility. Critics complain that the lack of specifics leaves organizations open to attack.

Nonetheless, this standard is one of the most valuable documents available to the information security industry. It bridges an often tangible gap between management and engineering and helps focus both parties on minimizing risk.

PCI DSS

The Payment Card Industry Data Security Standard (PCI DSS) is a different beast entirely. It arose from the perceived need to improve security in the credit card processing industry following a series of dramatic exposures. For example, in 2013, the U.S. government revealed the exposure of 160 million credit card numbers by various Visa licensees, including JCPenney. This is the largest cybercrime case in U.S. history; it’s estimated that more than $300 million was lost.
The PCI DSS standard is the result of a joint effort between Visa and MasterCard, though it is currently maintained by Visa. Unlike ISO 27001, it is freely available for anyone to download. It focuses entirely on protecting cardholder data systems and has 12 sections that define requirements for protection.

Because PCI DSS is focused on card processors, it is not generally appropriate for businesses that don’t deal with credit card data. However, for those that do, strict compliance is necessary to avoid hefty fines and possible criminal prosecution. You can find the document at pcisecuritystandards.org.

NIST 800 series

The fine folks at NIST have created the Special Publication (SP) 800 series of documents to report on their research, guidelines, and outreach efforts in computer security. These documents are most often used in connection with measuring FISMA (the Federal Information Security Management Act of 2002) compliance for those organizations that handle data for the U.S. federal government. More generally, they are publicly available standards with excellent content and have been widely adopted by industry.

The SP 800 series includes more than 100 documents. All of them are available from csrc.nist.gov/publications/PubsSPs.html. Table 27.3 lists a few that you might want to consider starting with.

Table 27.3: Recommended publications in the NIST SP 800 series

<table>
<thead>
<tr>
<th>Pub</th>
<th>Title</th>
<th></th>
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<tbody>
<tr>
<td>800-12</td>
<td>An Introduction to Computer Security: The NIST Handbook</td>
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<tr>
<td>800-14</td>
<td>Generally Accepted Principles and Practices for Securing IT Systems</td>
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<tr>
<td>800-34 R1</td>
<td>Contingency Planning Guide for Information Technology Systems</td>
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<tr>
<td>800-39</td>
<td>Managing Risk from Information Systems: An Organizational Perspective</td>
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<tr>
<td>800-53 R4</td>
<td>Recommended Security Controls for Federal IT and Organizations</td>
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<tr>
<td>800-123</td>
<td>Guide to General Server Security</td>
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</table>

The Common Criteria

The Common Criteria for Information Technology Security Evaluation (commonly known as the “Common Criteria”) is a standard for evaluating the security level of IT products. These guidelines were established by an international committee consisting of members from various manufacturers and industries. See commoncriteriaportal.org to learn more about the standard.

OWASP: the Open Web Application Security Project

OWASP is a nonprofit world-wide organization focused on improving the security of application software. It is best known for its “top 10” list of web application security risks, which helps
remind all of us where to focus our energies when securing applications. Find the current list and a bunch of other great material at owasp.org.

*CIS: the Center for Internet Security*

CIS has excellent resources for administrators. Perhaps the most valuable are the CIS benchmarks, a collection of technical configuration recommendations for securing operating systems. You can find benchmarks for each of our example UNIX and Linux systems. CIS also has benchmarks for cloud providers, mobile devices, desktop software, network devices, and more. Learn more at cisecurity.org.
27.11 SOURCES OF SECURITY INFORMATION

Half the battle of keeping your systems secure consists of staying abreast of security-related developments in the world at large. If your site is broken into, the break-in probably won’t happen through the use of a novel technique. More likely, the chink in your armor will turn out to have been a known vulnerability that has been widely discussed in vendor knowledge bases, on security-related newsgroups, and on mailing lists.
SecurityFocus.com, the BugTraq mailing list, and the OSS mailing list

SecurityFocus.com specializes in security-related news and information. The news includes current articles on general issues and on specific problems. The site also includes an extensive technical library of useful papers, nicely sorted by topic.

SecurityFocus’s archive of security tools contains software for a variety of operating systems along with blurbs and user ratings. It is the most comprehensive and detailed source of tools that we are aware of.

The BugTraq list is a moderated forum for the discussion of security vulnerabilities and their fixes. To subscribe, visit securityfocus.com/archive. Traffic on this list can be fairly heavy, however, and the signal-to-noise ratio is poor. A database of BugTraq vulnerability reports is also available from the web site.

The oss-security mailing list (openwall.com/lists/oss-security) is an excellent source of security tidbits from the open source community.
Schneier on Security

Bruce Schneier’s blog is a valuable and sometimes entertaining source of information about computer security, cryptography, and squid. Schneier is the author of the well-respected books *Applied Cryptography* and *Secrets and Lies*, among others. Information from the blog is also captured in the form of a monthly newsletter known as the Crypto-Gram. Learn more at [schneier.com/crypto-gram.html](http://schneier.com/crypto-gram.html).
The Verizon Data Breach Investigations Report

Released annually, this report is packed with statistics about the causes and sources of data breaches, and it’s an entertaining read to boot. The 2016 edition suggests, based on an analysis of 3,141 incidents, that around 80% of data breaches are financially motivated. Espionage comes in a distant second. This publication includes a useful breakdown of the types of attacks being seen in the wild.
The SANS Institute

The SANS (SysAdmin, Audit, Network, Security) Institute is a professional organization that sponsors security-related conferences and training programs, as well as publishing a variety of security information. Their web site, sans.org, is a useful resource that occupies something of a middle ground between SecurityFocus and CERT. It’s neither as frenetic as the former nor as stodgy as the latter.

SANS offers several weekly and monthly email bulletins that you can sign up for on their web site. The weekly NewsBites are nourishing, but the monthly summaries seem to contain a lot of boilerplate. Neither is a great source of late-breaking security news.
Distribution-specific security resources

Because security problems have the potential to generate a lot of bad publicity, vendors are usually eager to help customers keep their systems secure. Most large vendors have an official mailing list to which security-related bulletins are posted, and many maintain a web site about security issues as well. It’s common for security-related software patches to be distributed for free, even by vendors that normally charge for software support.

Security portals on the web, such as SecurityFocus.com, contain vendor-specific information and links to the latest official vendor dogma.

Ubuntu maintains a security mailing list at

https://lists.ubuntu.com/mailman/listinfo/ubuntu-security-announce

For Red Hat security information, subscribe to the “enterprise watch” list to get announcements about the security of Red Hat’s product line. Find it at

https://redhat.com/mailman/listinfo/enterprise-watch-list

Although CentOS advisories typically (always?) mirror Red Hat security advisories, it’s probably worthwhile to subscribe to the CentOS list at

https://lists.centos.org/pipermail/centos-announce/

FreeBSD has an active security group with a mailing list at

https://lists.freebsd.org/mailman/listinfo/freebsd-security
Other mailing lists and web sites

The contacts listed above are just a few of the many security resources available on the net. Given the volume of information that’s now available and the rapidity with which resources come and go, we thought it would be most helpful to point you toward some metaresources.

One good starting point is linuxsecurity.com, which logs several posts each day on pertinent Linux security issues. It also maintains a running collection of Linux security advisories, upcoming events, and user groups.

(IN)SECURE magazine is a free bimonthly magazine that includes news about current security trends, product announcements, and interviews with notable security professionals. Some of the articles should be read with a vial of salt nearby, and always check the bylines. In many cases, authors are just pimping their own products.

The Linux Weekly News is a tasty treat that includes regular updates about the kernel, security, distributions, and other topics. LWN’s security section can be found at lwn.net/security.
27.12 When your site has been attacked

The key to handling an attack is simple: don’t panic. It’s very likely that by the time you discover an intrusion, most of the damage has already been done. In fact, it has probably been going on for weeks or months. The chance that you’ve discovered a break-in that just happened an hour ago is slim to none.

In that light, the wise owl says to take a deep breath and begin developing a carefully thought out strategy for dealing with the break-in. You need to avoid tipping off the intruder by announcing the break-in or performing any other activity that would seem abnormal to someone who may have been watching your site’s operations for many weeks. Hint: performing a system backup is usually a good idea at this point and (hopefully!) will appear to be a normal activity to the intruder. (If system backups are not a “normal” activity at your site, you have much bigger problems than the security intrusion.)

This is also a good time to remind yourself that some studies have shown that 60% of security incidents involve an insider. Be very careful with whom you discuss the incident until you’re sure you have all the facts.

Here’s a quick 9-step plan that may assist you in your time of crisis:

1. **Don’t panic.** In many cases, a problem isn’t noticed until hours or days after it took place. Another few hours or days won’t affect the outcome. The difference between a panicky response and a rational response will. Many recovery situations are exacerbated by the destruction of important log, state, and tracking information during an initial panic.

2. **Decide on an appropriate level of response.** No one benefits from an overhyped security incident. Proceed calmly. Identify the staff and resources that must participate, and leave others to assist with the postmortem after it’s all over.

3. **Hoard all available tracking information.** Check accounting files and logs. Try to determine where the original breach occurred. Back up all your systems. Make sure that you physically write-protect removable media if you connect them to a live system.

4. **Assess your degree of exposure.** Determine what crucial information (if any) has “left” the company, and devise an appropriate mitigation strategy. Determine the level of future risk.

5. **Pull the plug.** If necessary and appropriate, disconnect compromised machines from the network. Close known holes and stop the bleeding. CERT recommends steps for analyzing an intrusion. The document can be found at cert.org/tech_tips/win-UNIX-system_compromise.html.
6. **Devise a recovery plan.** With a creative colleague, draw up a recovery plan on nearby whiteboard. This procedure is most effective when performed away from a keyboard. Focus on putting out the fire and minimizing the damage. Avoid assigning blame or creating excitement. In your plan, don’t forget to address the psychological fallout your user community may experience. Users inherently trust others, and blatant violations of trust make many folks uneasy.

7. **Communicate the recovery plan.** Educate users and management about the effects of the break-in, the potential for future problems, and your preliminary recovery strategy. Be open and honest. Security incidents are part of life in a modern networked environment. They are not a reflection on your ability as a system administrator or on anything else worth being embarrassed about. Openly admitting that you have a problem is 90% of the battle, as long as you can demonstrate that you have a plan to remedy the situation.

8. **Implement the recovery plan.** You know your systems and networks better than anyone. Follow your plan and your instincts. Speak with a colleague at a similar institution (preferably one who knows you well) to keep yourself on the right track.

9. **Report the incident to authorities.** If the incident involved outside parties, report the matter to CERT. They have a hotline at (412) 268-5800 and can be reached by email at cert@cert.org. Include as much information as you can.

A standard form is available from cert.org to help jog your memory. Here are some of the more useful pieces of information you might include:

- The names, hardware, and OS versions of the compromised machines
- The list of patches that had been applied at the time of the incident
- A list of accounts that are known to have been compromised
- The names and IP addresses of any remote hosts that were involved
- Contact information (if known) for the administrators of remote sites
- Relevant log entries or audit information

If you believe that a previously undocumented software problem may have been involved, report the incident to the software vendor as well.
27.13 Recommended Reading


A commitment to monitoring is the distinguishing characteristic of a professional system administrator. Inexperienced sysadmins often leave systems unmonitored and allow failures to be “detected” when a frustrated, angry user calls the help desk because they’re unable to complete an intended task. Slightly more clued-in administrative groups set up a monitoring platform but disable after-hours notifications because they are too bothersome. In either case, fire fighting and hilarity ensue. These approaches adversely affect the enterprise, complicate recovery efforts, and give the sysadmin team a bad reputation.

Professional sysadmins adopt monitoring as their religion. Every system is added to the monitoring platform before it goes live, and the battery of checks is regularly tested and tuned. Metrics and trends are evaluated proactively so that problems can be spotted before they affect users or put data at risk.

A major on-line video streaming service you may have heard of values their telemetry system so much that they’d rather have a service outage than a monitoring outage. Without monitoring, they’d have no idea what was happening anyway.

A monitoring-first philosophy (along with its associated tools) makes you a sysadmin superhero. You develop a better understanding of your software and applications, fix small problems before they snowball into catastrophic failures, and become more effective at finding error conditions, debugging problems, and understanding the performance of complex systems. Monitoring also
improves your quality of life by letting you fix most issues at your convenience rather than at 3:00 a.m. on Thanksgiving Day.
28.1 AN OVERVIEW OF MONITORING

The goals of monitoring are to ensure that the IT infrastructure as a whole operates as expected and to compile, in an accessible and easily digested form, data that are useful for management and planning. Simple, right? But this high-level description covers a potentially vast territory.

Real-world monitoring systems vary in every possible dimension, but they all share this same basic structure:

- Raw data is harvested from systems and devices of interest.
- The monitoring platform reviews the data and determines what actions are appropriate, usually by applying administratively set rules.
- The raw data and any actions decided on by the monitoring system flow through to back ends that take appropriate action.

Real-world monitoring systems range from trivially simple to arbitrarily complex. For example, the following Perl script includes all the elements listed above:

```perl
#!/usr/bin/env perl
$loadavg = (split /\s+/, 'uptime')[10];

# If load is greater than 5, notify sysadmin
if ($loadavg > 5.0) {
    system 'mail -s "Server load is too high" dan@admin.com < /dev/null'
}
```

The script runs the `uptime` command to obtain the system’s load averages. If the one-minute load average is larger than 5.0, it sends mail to an administrator. Data, evaluation, reaction.

Once upon a time, a “fancy” monitoring setup involved collections of scripts like this that ran from `cron` and commandeered a modem to send messages to sysadmins’ pagers. Today, you have multiple options available at every stage of the monitoring pipeline.

Of course, you can still write individual monitoring scripts and run them from `cron`. If this is really all you need, by all means keep things simple. But unless you are responsible for only one or two servers, this ad hoc approach is normally not sufficient.

The following sections review the stages of the pipeline in a bit more detail.
**Instrumentation**

A wide range of data that may prove useful to your organization includes performance figures (response time, utilization, transfer rate), availability figures (reachability and uptime), capacity, state changes, log entries, and even business metrics such as average shopping cart value or click conversion rate.

Because anything one might do on a computer is potentially of monitoring interest, monitoring systems are usually data-source agnostic. They often come with built-in support for a variety of inputs. Even data sources that lack direct support can normally be brought in with a few lines of adapter code or a separate data gateway such as StatsD (see [this page](#)).

With so much data out there begging to be collected, the hard part of designing a collection system can be knowing what to ignore. Avoid collecting data that does not have a clear and actionable purpose. Data overcollection loads down both the monitoring system and the entities being monitored. It also tends to obscure the values that are truly important, drowning them in a sea of noise.

Unfortunately, it’s often not easy to distinguish useful data from dross. You must continually reevaluate what is monitored and rethink how that data will be acted on throughout a system’s life.
Data types

At the highest level, monitoring data can be grouped into three general categories:

- **Real-time metrics**, which characterize the operational state of the environment. These are typically numbers or Boolean values. In general, it’s the responsibility of the monitoring system to test these metrics against expectations and generate an alert if a current value exceeds a predefined range or threshold.

- **Events**, which often take the form of log file entries or “push” notifications from subsystems. These events, sometimes known as pattern-based metrics, can indicate that a state change, alarm condition, or other action has occurred. Events can be processed to form numeric metrics (e.g., a total or a rate), or they can trigger monitoring responses directly.

Many of the data points collected by application monitoring software fall into the “event” category; sometimes they have quantitative data attached as well. Interrelationships among events (e.g., “the user looked at the Settings page but then canceled without changing anything”) are often helpful to investigate. General-purpose monitoring platforms tend not to be very good at this sort of cross-referencing, which is one reason that application monitoring is a category of its own.

- **Aggregated and summarized historic trends**, which are often time-series collections of real-time metrics. They allow for analysis and visualization of changes over time.
**Intake and processing**

Most monitoring systems revolve around a central monitoring platform that absorbs data from monitored systems, performs appropriate processing, and applies administrative rules to determine what should happen in response.

First-generation platforms such as Nagios and Icinga focused on detecting and responding to problems as they occurred. These systems were revolutionary for their day and led us into the modern world of monitoring. Nevertheless, they have been eclipsed over time by the industry’s gradual realization that all monitoring data is time-series data. If values didn’t vary, you wouldn’t be monitoring them.

Clearly, a more data-oriented approach was needed. However, monitoring data is usually so voluminous that you can’t simply dump it all into a traditional database and allow it to accumulate. That’s a recipe for poor performance and overflowing disks.

The modern approach is to organize monitoring around a data store that’s specialized for handling time-series data. All data is stored for an initial period, but as the data ages, the store applies increasingly high levels of summarization to limit storage requirements. For example, the store might keep an hour’s worth of data at one-second resolution, a week’s worth of data at one-minute resolution, and a year’s worth of data at one-hour resolution.

Historical data is useful not only for dashboard presentations, but also as a baseline for comparison. Is the current network error rate 25% or more above its historical average?
Notifications

Once you have a monitoring framework in place, put careful thought into what to do with the monitoring results. The first priority is usually to notify administrators and developers about a problem that needs attention.

Notifications must be actionable. Structure your monitoring system so that everyone who receives a given notification must potentially do something in response, even if the action is something as general as “check later to be sure this was taken care of.” Notifications that are purely informational train staff to ignore notifications.

In most cases, notifications need to extend beyond email to be optimally effective. For critical issues, SMS notifications (that is, text messages) to administrators’ cell phones are easy and efficient. Recipients can set their ring tones and phone volume so that they’ll be awakened in the middle of the night if desired.

See this page for more comments on ChatOps.

Notifications should also be integrated with your team’s ChatOps implementation. Less critical notifications (such as job statuses, login failures, and informational notices) can be sent to one or more chat rooms so that interested parties can actively receive subsets of alerts in which they might be interested.

Beyond these basic channels, the possibilities are endless. An LED lighting system that changes colors according to system status can be useful for at-a-glance status indication in a data center or network operations center, for example. Other options for responding to situations identified by the monitoring systems include

- Automated actions, such as dumping a database or rotating logs
- Calling an administrator on the phone
- Sending data to a wall board for public display
- Storing data in a time-series database for later analysis
- Doing nothing, and allowing later review through the system itself
**Dashboards and UIs**

Beyond alerting for clearly exceptional circumstances, one of the main goals of monitoring is to present the state of the environment in a manner that’s more structured and easier to assimilate than a bunch of raw data. Such displays are generically termed “dashboards.”

Dashboards are designed by administrators or by other stakeholders with an interest in particular aspects of the environment. They use several different techniques to transform raw data into infographic gold.

First, they’re selective in what they present. They concentrate on the most important metrics for a given domain, the ones that indicate general states of health or performance. Second, they give context clues to the significance and import of the data that’s shown. For example, problematic numbers and states are typically shown in red, and primary metrics are depicted in larger font sizes. Relationships among values are shown through grouping. Third, dashboards display data series as charts, making them easy to assess at a glance.

Of course, most data that’s collected never shows up on a dashboard. It’s helpful if your monitoring system also has a generalized UI that facilitates investigation and modification of the data schema, allows you to make arbitrary database queries, and charts arbitrarily defined sequences of data on the fly.
28.2 The Monitoring Culture

This chapter is mostly about tools, but culture is at least as important. When you embark on a monitoring journey, embrace the following tenets:

- If someone cares about or depends on a system or service, it must be monitored. Full stop. Nothing in the environment that a service or user depends on can remain unmonitored.

- If a production device, system, or service exposes monitorable attributes, those attributes should be monitored. Don’t let a server with a fancy “lights out” hardware management interface spend weeks futilely trying to notify you that a fan has failed.

- All high-availability constructs must be monitored. It would be unfortunate to learn that a primary server had failed only after the backup server failed, too.

- Monitoring is not optional. The work plans of every sysadmin, developer, ops staff member, manager, and project manager should include provisions for monitoring.

- Monitoring data (especially historical data) is useful to everyone. Make data easily accessible and visible so that everyone can use it to help with root cause analysis, planning, life cycle management, and architectural improvement opportunities. Put effort and resources into creating and promoting monitoring dashboards.

- Everyone should respond to alerts. Monitoring is not just an ops problem. All technical roles should receive notifications and work together to resolve issues. This approach encourages bona fide root cause analysis by whichever individuals are most suited to fix the underlying issue.

- Properly implemented, monitoring impacts quality of life in a positive way. A solid monitoring regimen frees you from the burden of worrying about what state your systems are in and empowers others to support you. Without monitoring and appropriate documentation, you are essentially on call 24 × 7 × 365.

- Train responders to fix alerts, not just suppress them. Evaluate false-positive or noisy alerts and tune them so that they no longer trigger inappropriately. Spurious alerts encourage everyone to ignore the monitoring system.
28.3 The Monitoring Platforms

If you plan to monitor multiple systems and more than a few metrics, it’s worth investing some time into the deployment of a full-service monitoring platform. These are general-purpose systems that collect data from multiple sources, facilitate the display and summarization of status information, and establish a standard way to define actions and alerts.

The good news is that there are a variety of choices. The not-as-good news is that no single, perfect platform as yet exists. When selecting from among the available options, consider the following issues:

- **Data-gathering flexibility.** All platforms can absorb data from a variety of sources. However, that doesn’t mean that all platforms are equivalent in this regard. Consider the data sources you want to actually use. Will you need to read data from an SQL database? From DNS records? From an HTTP connection?

- **User interface quality.** Many systems offer customizable GUIs or web interfaces. Most well-marketed packages today tout their ability to understand JSON templates for data presentation. A UI is not just marketing hype; you need an interface that relays information clearly, simply, and comprehensibly. Will you need different user interfaces for different groups within your organization?

- **Cost.** Some commercial management packages come at a stiff price. Many corporations find value in being able to say that their site is managed by a high-end commercial system. If that isn’t so important to your organization, look at free options such as Zabbix, Sensu, Cacti, and Icinga.

- **Automated discovery.** Many systems offer to “discover” your network. Through a combination of broadcast pings, SNMP requests, ARP table lookups, and DNS queries, they identify all your local hosts and devices. All the discovery implementations we have seen work pretty well, but accuracy is lower on complex or heavily firewalled networks.

- **Reporting features.** Many products can send alert email, integrate with ChatOps, send text messages, and automatically generate tickets for popular trouble-tracking systems. Make sure that the platform you choose accommodates flexible reporting. Who knows what electronic devices you’ll be dealing with in a few years?
Open source real-time platforms

Although the platforms in this section—Nagios, Icinga, and Sensu Core—do a little bit of everything, they’re known for their strength in handling instantaneous (or threshold-based) metrics.

These systems have their proponents, but as first-generation monitoring tools, they’re gradually losing favor to time-series systems, which we describe starting here. Most sites starting from scratch would be better advised to opt for a time-series system.

Nagios and Icinga

Nagios and Icinga specialize in real-time notification of error conditions. Although they do not help you determine how much your bandwidth utilization has increased over the last month, they can track you down when your web server goes off-line.

Nagios and Icinga were originally forks of a single source tree, but modern-day Icinga 2 has been completely rewritten. However, it remains compatible with Nagios in most respects.

Both systems include scores of scripts for monitoring services of all shapes and sizes, along with extensive SNMP monitoring capabilities. Perhaps their greatest strength is their modular and heavily customizable configuration system, which allows you to write custom scripts to monitor any conceivable metric.

You can whip up new monitors in Perl, PHP, Python, or even C if you’re feeling ambitious and masochistic. Many standard notification methods are built in—email, web reports, text messages, etc. And as with monitoring plug-ins, it’s easy to roll your own notification and action scripts.

Nagios and Icinga both work well for networks of fewer than a thousand hosts and devices. They are easy to customize and extend, and include powerful features such as redundancy, remote monitoring, and notification escalation.

If you are deploying new monitoring infrastructure from scratch, we recommend Icinga 2 over Nagios. Its code base is generally cleaner, and it has been rapidly accreting fans and community support. From a functional perspective, its UI is cleaner and faster, and it’s able to autobuild service dependencies, which can be essential in complex environments.

Sensu

Sensu is a full-stack monitoring framework that’s available both as an open source edition (Sensu Core) and with paid, commercially supported add-ons. It has an ultramodern UI and can run any legacy Nagios, Icinga, or Zabbix monitoring plug-in. It was designed as a replacement for Nagios, so plug-in compatibility is one of its most attractive features. Sensu allows for easy integration with Logstash and Slack notifications, and its installation process is particularly easy.
Open source time-series platforms

Detecting and responding to current problems is just one aspect of monitoring. It’s often equally important to know how values are changing over time and how they relate to other values. Four popular time-series platforms aim to scratch this itch: Graphite, Prometheus, InfluxDB, and Munin.

These systems put the database front and center within the monitoring ecosystem. They vary in their degree of completeness as stand-alone monitoring systems, and in general are designed for a more modular world than traditional systems such as Icinga. You may need to supply some additional components to build a complete monitoring platform.

Graphite

Graphite was arguably the vanguard of the new generation of time-series monitoring platforms. At its core is a flexible time-series database with an easy-to-use query language. The reason for the #monitoringlove movement and for the enormous influence that Graphite has had on front-end UIs is the way that it aggregates and summarizes metrics. It started the move away from per-minute monitoring and toward sub-second monitoring.

As you might guess from the name, Graphite includes graphing features for web visualization. However, this aspect of the package has been somewhat eclipsed by Grafana. Graphite is better known these days for its other components, Carbon and Whisper, which form the core of the data management system.

Graphite can be combined with other tools to create a scalable, distributed, clustered, monitoring environment that is capable of absorbing and reporting on hundreds of thousands of metrics. Exhibit A shows an architectural diagram of such an implementation.

Exhibit A: Clustered Graphite architecture
Prometheus

Our favorite time-series platform today is Prometheus. It’s a comprehensive platform that includes integrated collection, trending, and alerting components. The components are both sysadmin- and developer-friendly, which makes it a great choice for a DevOps shop. It does not allow for clustering, however, which may mean that it’s not the right fit for sites that require high availability.

InfluxDB

InfluxDB is an extraordinarily developer-friendly time-series monitoring platform that supports a broad array of programming languages. Much like Graphite, InfluxDB is really just a time-series database engine. You’ll need to complete the package with external components such as Grafana to form a complete monitoring system that includes features like alerting.

The data management features of InfluxDB are much richer than those of the alternatives listed above. However, InfluxDB’s additional features also add some unwelcome complexity for typical installations.
InfluxDB has had a somewhat troubled history of bugs and incompatibilities. However, the current version appears to be stable and is probably the best current alternative to Graphite if you are seeking a stand-alone data management system.

*Munin*

Munin has historically been quite popular, especially in Scandinavia. It’s built on a clever architecture in which the data collection plug-ins not only provide data but also tell the system how the data should be presented. Although Munin is still perfectly usable, modern alternatives such as Prometheus should be considered for new deployments. Munin is still a useful tool for application-specific monitoring in some cases; see [this page](#).
Open source charting platforms

The two main choices for creating dashboards and charts are the graphing features built into Graphite and a newer package, Grafana.

Graphite can draw data from stores other than Whisper (the native data-storage component of the Graphite package), but this isn’t necessarily a well-trodden path.

As a database-agnostic package, Grafana does quite well at accommodating foreign data stores, including all those listed in the previous section. At last count, more than 30 back ends were supported. Grafana originally started out as an attempt to improve graphing for Graphite, so it’s quite comfortable in a Graphite environment, too.

Both Graphite and Grafana present a dashboard-like graphing interface that can generate insight-provoking and management-pleasing visualizations. You can use them to display anything from low-level system metrics to business-level indicators. Bake-offs usually give the nod to Grafana for its superior UI and prettier graphs.

Exhibit B shows a simple Grafana dashboard.

Exhibit B: Grafana dashboard example

![Grafana dashboard example](image_url)
Commercial monitoring platforms

Hundreds of companies sell monitoring software, and new competitors enter the market every week. If you are looking for a commercial solution, you should at least consider the options listed in Table 28.1.

Table 28.1: Popular commercial monitoring platforms

<table>
<thead>
<tr>
<th>Platform</th>
<th>URL</th>
<th>Comments</th>
</tr>
</thead>
<tbody>
<tr>
<td>Datadog</td>
<td>datadoghq.com</td>
<td>Cloud-based application monitoring platform</td>
</tr>
<tr>
<td></td>
<td></td>
<td>Huge list of supported systems, apps, and services</td>
</tr>
<tr>
<td>Librato</td>
<td>librato.com</td>
<td>Plug and Play with existing open source plugins</td>
</tr>
<tr>
<td>Monitus</td>
<td>monitus.net</td>
<td>E-commerce platform monitoring</td>
</tr>
<tr>
<td>Pingdom</td>
<td>pingdom.com</td>
<td>SaaS-based monitoring platform(^a)</td>
</tr>
<tr>
<td>SignalFx</td>
<td>signalfx.com</td>
<td>SaaS platform with long list of cloud integrations</td>
</tr>
<tr>
<td>SolarWinds</td>
<td>solarwinds.com</td>
<td>Network monitoring stalwart</td>
</tr>
<tr>
<td>Sysdig Cloud</td>
<td>sysdig.com</td>
<td>Specialty: Docker monitoring and alerting</td>
</tr>
<tr>
<td></td>
<td></td>
<td>Easy to correlate events across services</td>
</tr>
<tr>
<td>Zenoss</td>
<td>zenoss.com</td>
<td>Incredibly complex alternative to Icinga</td>
</tr>
</tbody>
</table>

\(^a\) No software install required. Good fit for web apps only.

Whether their systems reside in the cloud, on a data center hypervisor, or in a closet, most businesses should not be building their own monitoring stack. Outsourcing is cheaper and more reliable. Hence, consider Datadog, Librato, SignalFx, or Sysdig Cloud if you need a monitoring stack for a common set of applications or servers.

When investigating commercial monitoring platforms, you often first consider price. But don’t forget to research the operational details as well:

- How easy is it to integrate into your configuration management system?
- How does the system deploy new plug-ins or checks to your hosts? Are they pushed or pulled?
- Does it integrate well with your existing notification platform if you have one?
- Does your environment allow the type of external connectivity needed to facilitate a cloud-based monitoring solution?

These are just a few of the questions you should be asking when researching platforms. In the end, the best platform for your site is one that is easily configurable, meets your budget, and is easily adopted by your users.
Hosted monitoring platforms

If you’re not interested in setting up and maintaining your own network monitoring tools, you might want to consider a hosted (cloud) solution. Many free and commercial options exist, but a popular one is StatusCake, statuscake.com. An external provider’s ability to see the internal details of your network is limited, but hosted options work well for validating the health of public-facing services and web sites.

A hosted monitoring provider can also liberate you from the constraints of your organization’s normal Internet connection. If you rely on your upstream network to transport notifications from an internal monitoring system—as most sites ultimately do—you may want to ensure that your upstream network is itself monitored and instrumented so that staff can be marshaled in the event of trouble.
28.4 Data collection

The previous sections reviewed a variety of packages that can serve as a site’s central monitoring engine. However, selecting and deploying one of these systems is only the first part of the setup process. You must now make sure that the data and events you’re interested in monitoring make their way into the central monitoring platform.

The details of this instrumentation process depend on the systems you want to monitor and the philosophy of your monitoring platform. In many cases, you’ll need to write some simple glue scripts to convert status information into a form that your monitoring platform can understand. Some platforms, such as Icinga, come with a wide variety of plug-ins that harvest standard metrics from commonly monitored systems. Others, such as Graphite and InfluxDB, make no real provision for data input at all and must be supplemented by a front end that handles this role.

In the following sections, we first look at StatsD, a general-purpose data collection front end, then review some tools and techniques for instrumenting some commonly monitored systems.
StatsD: generic data submission protocol

StatsD was written by engineers at Etsy as a way to track anything and everything within their own environment. It’s a UDP-based front-end proxy that dumps any data you throw at it into a monitoring platform for consumption, calculation, and display. StatsD’s superpower is its ability to ingest and perform calculations on arbitrary statistics.

Etsy’s StatsD daemon was written in Node.js. But these days, “StatsD” refers more to the protocol than to any one of the many software packages that implement it. (Truth be told, even Etsy’s version is not the original; it was inspired by a similarly named project at Flickr.) Implementations have been written in many different languages, but we focus on the Etsy release here.

StatsD depends on Node.js, so make sure that package has been installed and configured appropriately before you move on to installing StatsD. The Etsy implementation isn’t included in most OS vendors’ package repositories, though other versions of StatsD often are; make sure you don’t confuse them. It’s easiest to clone the Etsy version directly from GitHub:

$ git clone https://github.com/etsy/statsd

StatsD is incredibly modular and can feed the incoming data to a variety of back ends and clients. Let’s look at a simple example that uses Graphite as the back end.

To ensure that Graphite and StatsD communicate correctly, you must modify Carbon, Graphite’s storage component. Edit /etc/carbon/storage-schemas.conf and add a stanza similar to the following:

[stats]
pattern = ^stats.*
retentions = 10s:12h,1min:7d,10min:1y

This configuration tells Carbon to keep 12 hours of data at 10-second intervals. Carbon summarizes expiring data at 1-minute intervals and keeps that summary information for an additional 7 days. Similarly, data at 10-minute granularity is maintained for a full year. There’s nothing magic about these choices; you’ll need to determine what’s appropriate for your organization’s retention needs and the data being collected.

The exact definition of what it means to “summarize” time-series data varies according to the type of data. If you’re counting network errors, for example, you probably want to summarize by adding up values. If you’re looking at metrics that represent load or utilization, you probably want an average. You might also need to specify appropriate ways of handling missing data.

These policies are specified in the file /etc/carbon/storage-aggregation.conf. If you don’t already have a working Graphite installation, you might find Graphite’s sample configuration useful as a starting point:
Below are some reasonable defaults to include in the `storage-aggregation.conf` file.

```
[min]
pattern = \.lower$
xFilesFactor = 0.1
aggregationMethod = min

[max]
pattern = \.upper(_\d+)?$
xFilesFactor = 0.1
aggregationMethod = max

[sum]
pattern = \.sum$
xFilesFactor = 0
aggregationMethod = sum

[count]
pattern = \.count$
xFilesFactor = 0
aggregationMethod = sum

[count_legacy]
pattern = ^stats_counts.*
xFilesFactor = 0
aggregationMethod = sum

[default_average]
pattern = .*
xFilesFactor = 0.3
aggregationMethod = average
```

Note that every configuration block has a regular expression `pattern` that attempts to match the names of data series. Blocks are read in order, and the first matching block becomes the controlling specification for each data series. For example, a series named `sample.count` would match the pattern for the `[count]` block. The values would be rolled up by adding up data points (`aggregationMethod = sum`).

The `xFilesFactor` setting determines the minimum number of samples needed to meaningfully downsample your metric. It’s expressed as a number between 0 and 1 that represents the percentage of non-null values that must exist at the more granular layer in order for the rollup layer to have a non-null value. For example, the `xFilesFactor` setting for `[min]` and `[max]` above is 10%, so even a single data value will satisfy this criterion, given our settings in the `storage-schema.conf` file. The default is 50%. If the numbers aren’t thoughtfully set, you’ll get inaccurate or missing data!

We can send some test data to StatsD with netcat (`nc`):
$ echo "sample.count:1|c" | nc -u -w0 statsd.admin.com 8125

This command submits a value of 1 as a count metric (as indicated by the `c`) to the sample.count data set. The packet goes to port 8125 on statsd.admin.com; this is the port on which `statsd` listens by default. If this datum shows up in your Graphite dashboard, you’re ready to collect all kinds of monitoring data through one of the many StatsD clients. See the StatsD GitHub wiki page (github.com/etsy/statsd/wiki) for a list of some of the clients that can communicate with StatsD. Or write your own! The protocol is simple and the possibilities are endless.
Data harvesting from command output

If you can investigate something from the command line, you can track it in your monitoring platform. All you need is a few lines of scripting glue to extract the data nuggets you’re interested in, which you then massage into a format your monitoring platform can accept.

For example, `uptime` shows the length of time that the system has been up, the number of logged-in users, and the load averages over the last 1, 5, and 15 minutes.

```
$ uptime
07:11:50 up 22 days, 10:13, 2 users, load average: 1.20, 1.41, 1.88
```

As a human, you can parse the output at a glance and see that the current load average is 1.20. If you want to write a script to check that value regularly or to feed it to another monitoring process, you can use text manipulation commands to isolate the desired value:

```
$ uptime | perl -anF'[^,]+' -e 'print $F[10]'
1.20
```

Here, we use Perl to split the output wherever there’s a sequence of spaces and commas and to print the contents of the tenth field (the 1-minute load average). Voilà!

Although Perl has been eclipsed by modern languages like Python and Ruby in most domains, it remains the king of quick-and-dirty text wrangling. It’s probably not worth learning Perl solely for this use, but Perl’s ability to phrase sophisticated text transformations as one-line commands does come in handy.

We can easily expand this one-liner into a short script that determines the load average and submits it to StatsD:

```
#!/usr/bin/env perl
use Net::Statsd;
use Sys::Hostname;
$Net::Statsd::HOST = 'statsd.admin.com';
$loadavg = (split /\s+/, 'uptime')[10];
Net::Statsd::gauge(hostname . '.loadAverage' => $loadavg);
```

Compare this script with our one-line StatsD test command from this page and our one-line parsing of `uptime` output above. Here, Perl has to run the `uptime` command and process its output as a string, so that portion looks somewhat different from its one-liner equivalent. (The one-liner relies on Perl’s autosplit mode.)

Instead of using `nc` to handle the network transmission of data to StatsD, we use a simple StatsD wrapper that we downloaded from the Comprehensive Perl Archive Network at cpan.org. That’s
generally the preferred approach; libraries are less brittle than hacks, and they clarify the code’s intent.

Many commands can generate more than one output format. Check the man page for the command to see what options are available before you attempt to parse its output. Some formats are much easier to deal with than others.

A few commands support an output format that specifically facilitates parsing. Others have configurable output systems in which you ask for only the fields that you really want. Yet another common option is a flag that suppresses descriptive header lines in the output.
28.5 Network Monitoring

Network status monitoring has traditionally been many sites’ first foray into the wider world of monitoring and dashboards, so it’s the first of several types of monitoring that we look at in a bit more depth. In subsequent sections, we also look at OS-level monitoring, application and service monitoring, and security monitoring.

The basic unit of network monitoring is the network ping, also known as an ICMP Echo Request packet. We discuss the technical details more thoroughly starting here, along with the ping and ping6 commands, which initiate pings from the command line.

The concept is simple: you send an echo request packet to another host on the network, and that host’s IP implementation returns a packet to you in response. If you receive a response to your probe, you know that all the network gateways and devices that lie between you and the target host are operational. You also know that the target host is powered on and that its kernel is up and running. However, because pings are handled within the TCP/IP protocol stack, they don’t guarantee anything about the state of higher-level software that might be running on the target host.

Pings don’t impose much overhead on the network, so it’s OK to send them frequently; say, every ten seconds. Design your pinging strategy thoughtfully, so that it covers all important gateways and networks. Keep in mind that if a ping can’t get through a gateway, neither can monitoring data that reports the failure of pings. You’ll want at least one set of pings to originate on the central monitoring host itself.

Network gateways aren’t required to answer ping packets, so pings might be dropped by a busy gateway. Even a properly functioning network loses a packet now and then. Ergo, don’t set off alarms at the first sign of trouble. It makes sense to collect ping data as binary event records (got through/didn’t get through) and roll it up into aggregate measures of percentage packet loss over longer terms.

You might also find it interesting to measure throughput between two points on the network. That can be done with iPerf; see this page for details.

Most network devices support the Simple Network Management Protocol (SNMP), an industry-standard way of naming and collecting operational data. Although SNMP has metastasized far beyond its networking roots, we consider it obsolete for purposes other than basic network monitoring.

SNMP is a rather large topic of its own, so we defer further discussion of it until later in this chapter. See SNMP: the Simple Network Management Protocol for details.
28.6 Systems Monitoring

Since the kernel controls a system’s CPU, memory, I/O, and devices, most of the interesting system-level state information you might want to monitor lives somewhere inside the kernel. Whether you’re investigating a particular system by hand or setting up an automated monitoring platform, you need the right tools to extract and expose this state information. Most kernels define formal channels through which such information is exported.

Unfortunately, kernels are like other types of software; error checking, instrumentation, and debugging features are often something of an afterthought. Although recent years have brought improvements in transparency, identifying and understanding the exact parameter you might want to monitor can be challenging and sometimes impossible.

See [this page](#) for more information about the `/proc` filesystem.

A particular value can often be obtained in more than one way. In the case of load averages, for example, you can read the values directly from `/proc/loadavg` on Linux systems or with `sysctl -n vm.loadavg` on FreeBSD. Load averages are also included in the output of the `uptime`, `w`, `sar`, and `top` commands (though `top` would be a poor choice for noninteractive use). It’s generally easiest and most efficient to access values directly from the kernel (through `sysctl` or `/proc`) if you can.

Monitoring platforms such as Nagios and Icinga include a rich set of community-developed monitoring plug-ins that you can use to get your hands on commonly monitored elements. They, too, are often simply scripts that run commands and parse the resulting output, but they come already tested and debugged, and they often work on multiple platforms. If you can’t find a plug-in that yields the value you’re interested in, you can write your own.
Commands for systems monitoring

Table 28.2 lists some command that are commonly used in monitoring. Many of these commands yield wildly different output depending on the command-line options you supply, so check the man pages for details.

Table 28.2: Commands that yield commonly monitored parameters

<table>
<thead>
<tr>
<th>Command</th>
<th>Available information</th>
</tr>
</thead>
<tbody>
<tr>
<td>df</td>
<td>Free and used disk space and inodes</td>
</tr>
<tr>
<td>du</td>
<td>Directory sizes</td>
</tr>
<tr>
<td>free</td>
<td>Free, used, and swap (virtual) memory</td>
</tr>
<tr>
<td>iostat</td>
<td>Disk performance and throughput</td>
</tr>
<tr>
<td>lsof</td>
<td>Open files and network ports</td>
</tr>
<tr>
<td>mpstat</td>
<td>Per-processor utilization on multiprocessor systems</td>
</tr>
<tr>
<td>vmstat</td>
<td>Process, CPU, and memory statistics</td>
</tr>
</tbody>
</table>

The Swiss Army knife of command-line data extraction is sar (short for “system activity report”). This command has a sordid history, having originally been introduced in System V UNIX in the 1980s. Old-school sysadmins are often identifiable by their fluency in sar.

The primary attraction of this command is that it has been implemented on a wide variety of systems, so it enhances the portability of both scripts and sysadmins. Sadly, the BSD port is no longer maintained.

The example below requests reports every two seconds for a period of one minute (i.e., 30 reports). The DEV argument is a literal keyword, not a placeholder for a device or interface name.

```
$ sar -n DEV 2 30
17:50:43 IFACE rxpck/s txpck/s rxbyt/s txbyt/s rxcmp/s txcmp/s rxmcsr/s
17:50:45 lo 3.61 3.61 263.40 263.40 0.00 0.00 0.00
17:50:45 eth0 18.56 11.86 1364.43 1494.33 0.00 0.00 0.52
17:50:45 eth1 0.00 0.00 0.00 0.00 0.00 0.00 0.00
```

This example is from a Linux machine with two network interfaces. The output includes both instantaneous and average readings of interface utilization in units of both bytes and packets. The second interface (eth1) is clearly not in use.
collectd: generalized system data harvester

As the work of system administration has evolved from wrangling individual systems to managing fleets of virtualized instances, simple command-line tools have started to create a lot of friction in the monitoring world. Although writing scripts to collect and analyze parameters is a utilitarian and flexible approach, maintaining the consistency of that code base across multiple systems quickly becomes cumbersome. Modern tools such as collectd, sysdig, and dtrace offer a more scalable approach to collecting this type of data.

Collection of system statistics should be a continuous process, and the UNIX solution to an ongoing task is to create a daemon to handle it. Enter collectd, the system statistics collection daemon.

This popular and mature tool runs on both Linux and FreeBSD. Typically, collectd runs on the local system, collects metrics at specified intervals, and stores the resulting values. You can also configure collectd to run in client/server mode, where one or more collectd instances aggregate data from a group of other servers.

Specification of the metrics to be collected and the destinations to which they are saved is flexible; over 100 plug-ins are available to suit your exact needs. Once collectd is running, it can be queried by a platform such as Icinga or Nagios for instantaneous monitoring or can forward data to a platform such as Graphite or InfluxDB for time-series analysis.

An example collectd configuration file is shown below.

```bash
## /etc/collectd/collectd.conf
Hostname client1.admin.com
FQDNLookup false
Interval 30
LoadPlugin syslog
  <Plugin syslog>
    LogLevel info
  </Plugin>

LoadPlugin cpu
LoadPlugin df
LoadPlugin disk
LoadPlugin interface
LoadPlugin load
LoadPlugin memory
LoadPlugin processes
LoadPlugin rrdtool
  <Plugin rrdtool>
    DataDir "/var/lib/collectd/rrd"
  </Plugin>
```
This basic configuration collects a variety of interesting system statistics every 30 seconds and writes RRDtool-compatible data files in /var/lib/collectd/rrd.
sysdig and dtrace: execution tracers

sysdig (Linux) and dtrace (BSD) comprehensively instrument both kernel and user process activity. They include components that are inserted into the kernel itself, exposing not only deep kernel parameters but also per-process system calls and other performance statistics. These tools are sometimes described as “Wireshark for the kernel and processes.”

See this page for more information about Wireshark.

Both of these tools are complex. However, they are well worth tackling. A weekend spent learning either one will give you amazing new superpowers and ensure your status as an A-list guest on the sysadmin cocktail circuit.

See Chapter 25 for more information about Docker and containers.

sysdig is container-aware, so it confers extraordinary visibility into environments where tools such as Docker and LXC are in use. sysdig is distributed as open source, and you can integrate it with other monitoring tools such as Nagios or Icinga. The developers also offer a commercial monitoring service (Sysdig Cloud) that has full monitoring and alerting capability.
28.7 Application Monitoring

At the top of the software ziggurat, we find the holy grail: application monitoring. This type of monitoring is rather vaguely defined, but the general idea is that it attempts to validate the status and performance of particular pieces of software rather than systems or networks as a whole. In many cases, application monitoring can reach into those systems and profile their internal operations.

To make sure you’re monitoring the right things, you need business units and developers to join the party and tell you more about their interests and concerns. If you have a web site that runs on the LAMP stack, for example, you’ll probably want to make sure you’re monitoring page load times, flagging critical PHP errors, keeping tabs on the MySQL database, and monitoring for specific issues such as excessive connection attempts.

Although monitoring for this layer can be complex, this domain is also where monitoring gets sexy. Imagine monitoring (and pulling into your beautiful Grafana dashboard) the number of widgets you’ve sold in the past hour or the average length of time that an item stays in a shopping cart. If you show your application developers and process owners that level of functionality, you usually get immediate buy-in to add more monitoring and may even get some help implementing it. Eventually, this layer of monitoring becomes invaluable to the business, and you start to be viewed as the champion of monitoring, metrics, and data analysis.

Application-level monitoring can yield additional insight into other events within your environment. For example, if widget sales decrease quickly, that might be an indication that one of your advertisement networks is down.
Log monitoring

In its most basic form, log monitoring involves grepping through log files to find interesting data you’d like to monitor, pulling out that data, and processing it into a form that’s usable for analysis, display, and alerting. Since log messages consist of free-form text, implementation of this pipeline can range in complexity from trivial to challenging.

Logs are typically best managed with a comprehensive aggregation system designed for that purpose. We address such systems in the section Management of logs at scale. Although these systems focus primarily on centralizing log data and making it easy to search and review, most aggregation systems also support threshold, alarm, and reporting functionality.

If you need automated log review for a few specific purposes and are reluctant to commit to a more general log management solution, we recommend a couple of smaller-scale tools: logwatch and OSSEC.

logwatch is a flexible, batch-oriented log summarizer. Its primary use is to create daily summaries of events reported in the logs. You can run logwatch more often than once a day, but it isn’t designed for real-time monitoring. For that, you would probably want to take a look at OSSEC, which we discuss here. OSSEC is promoted as a security tool, but its architecture is general enough that it’s useful for other kinds of monitoring as well.
Supervisor + Munin: a simple option for limited domains

An all-singing, all-dancing platform such as Icinga or Prometheus might be overkill for your needs or your environment. What if you’re only interested in monitoring one particular application process and don’t want the headache of a full-fledged monitoring platform? Consider combining Munin and Supervisor. They’re easy to install, require little configuration, and work well together.

Supervisor and its server process supervisord help you monitor processes and generate events or notifications when the processes exit or throw an exception. The system is similar in spirit to Upstart or to the process-management portions of systemd.

As mentioned on this page, Munin is a general monitoring platform with particular strengths in application monitoring. It’s written in Perl and requires an agent known as a Munin Node to be running on all the systems you want to monitor. Setting up a new Node is easy: just install the munin-node package, edit munin-node.conf to point to the master machine, and you’re good to go.

Munin defaults to creating RRDtool graphs with the data that it collects, so it’s a nice way to get some graphical feedback without much configuration. More than 300 plug-ins are distributed with Munin, and nearly 200 others are available as contributed libraries. It’s likely that you can find an existing plug-in that meets your needs. If not, it’s easy to write a new script for munin-node to execute.
Commercial application monitoring tools

If you search Google for “application monitoring tool,” you’ll discover many pages of commercial offerings to evaluate. For gold-star due diligence, you’ll also need to scrub through layers of recent discussions regarding application performance monitoring (APM).

See this page for more information about DevOps.

You’ll see many references to DevOps in these venues, and for good reason: application monitoring and APM are key tenets of DevOps. They supply quantitative metrics that teams can use to decide which areas of the stack would most benefit from efforts to improve performance and stability.

We think New Relic (newrelic.com) and AppDynamics (appdynamics.com) are standouts in this area. These systems’ capabilities overlap in many ways, but AppDynamics typically targets more a “full stack” monitoring solution, whereas New Relic deals more with profiling behavior inside the application layer itself.

Regardless of how you monitor your applications, it’s crucial to keep the development team involved in the process. They’ll help ensure that all important metrics are being monitored. Close cooperation on monitoring fosters the relationship between teams and limits duplication of effort.
28.8 Security monitoring

Security monitoring is a universe of its own. This area of operational practice is sometimes known as security operations or SecOps.

Dozens of open source and commercial tools and services can be enlisted to help monitor an environment’s security. Third parties, sometimes called managed security service providers (MSSPs), render outsourced services. Despite all these options, security breaches remain common and often go undetected for months or years.

It always sounds attractive to outsource security operations; then it becomes someone else’s problem to make sure your environment is secure. But think of it this way: would you be comfortable paying someone to watch your cash-filled wallet sit on a table with 10,000 other wallets in a busy train station? If so, an MSSP might be a good fit for you!

Perhaps the most important thing to know about security monitoring is that no automated tool or service is enough. You must implement a comprehensive security program that includes standards for user behavior, data storage, and incident response procedures, just to name a few elements. Chapter 27, Security, covers these basics.

Two core security functions should be integrated with your automated, continuous monitoring strategy: system integrity verification and intrusion detection.
System integrity verification

System integrity verification (often called file integrity monitoring or FIM) is the validation of the current state of the system against a known-good baseline. Most often, this validation compares the contents of the system files (kernel, executable commands, config files) with a cryptographically sound checksum such as SHA-512. If the checksum value of the file in the running system is different from that of the baseline version, a sysadmin is notified. Of course, regular maintenance activities such as planned changes, updates, and patches must be taken into account; not all changes are suspicious.

Acceptable hashing algorithms change over time. For example, MD5 is no longer considered cryptographically secure and should no longer be used.

The most commonly deployed FIM platforms are Tripwire and OSSEC; the latter is described in more detail starting here. The Linux version of AIDE also includes file integrity monitoring, but unfortunately, the FreeBSD version lacks this component.

Simpler is often better. A great bare-bones FIM option is mtree, which is native to FreeBSD and has recently been ported to Linux. mtree is an easy way to monitor file state and content changes, and it’s easily integrated into your monitoring scripts. Here’s an example of a quick script that uses mtree:

```
#!/bin/bash
if [ $# -eq 0 ]; then
    echo "mtree-check.sh [-bv]"
    echo "-b = create baseline"
    echo "-v = verify against baseline"
    exit
fi

## seed
KEY=93948764681464

## baseline directory
DIR=/usr/local/lib/mtree-check
if [ $1 = "-b" ]; then
    rm -rf $DIR/mtree_*
    cd $DIR
    mtree-c -K sha512 -s $KEY -p /sbin > mtree_sbin
fi

if [ $1 = "-v" ]; then
    cd $DIR
    mtree -s $KEY -p /sbin < mtree_sbin | \
        mail -s "\'hostname\' mtree integrity check" dan@admin.com
fi
```
With the `-b` flag, this script creates and stores the baseline. When run again with the `-v` flag, it validates the current contents of `/sbin` against the baseline.

As with so many aspects of system administration, setting up a FIM platform and operating it over time are wildly different propositions. You’ll need a defined process in place to maintain the FIM data and respond to FIM alerts. We suggest feeding information from your FIM platform into your monitoring and alerting infrastructure so that it is not sidelined or ignored.
Intrusion detection monitoring

Two common forms of intrusion detection systems (IDSs) are in use: host-based (HIDS) and network-based (NIDS). NIDS systems examine the traffic transiting the network and attempt to identify unexpected or suspicious patterns. The most common NIDS system is based on Snort and is covered in detail starting here.

HIDS systems run as a set of processes on each system. They typically keep tabs on a variety of things, including network connections, file modification times and checksums, daemon and applications logs, use of elevated privileges, and other clues that may signal the operation of tools designed to facilitate unauthorized access (“root kits”). A HIDS is not a one-stop solution for security, but it’s a valuable component of a comprehensive approach.

The two most popular open source HIDS platforms are OSSEC (Open Source SECurity) and AIDE (the Advanced Intrusion Detection Environment). In our experience, OSSEC is hands-down the better choice. Although AIDE is a nice FIM platform on Linux, OSSEC includes a broader array of functions. It can even be used in a client/server mode that supports non-UNIX clients such as Microsoft Windows and a variety of network infrastructure devices.

Much like FIM alerts, HIDS data is only as useful as the attention that’s paid to it. HIDS is not a “set it and forget it” subsystem, and you will need to integrate HIDS alerts with your overall monitoring system. The most effective strategy we’ve found for addressing this issue is to auto-open tickets for HIDS alerts in your trouble ticketing system. You can then add a monitoring check that alerts on any unresolved HIDS tickets.
28.9 SNMP: the Simple Network Management Protocol

Years ago, the networking industry decided it would be helpful to create a standard protocol for the collection of monitoring data. Hence, the Simple Network Management Protocol, aka SNMP.

Despite its name, SNMP is actually quite complex. It defines a hierarchical namespace of management data and methods for reading and writing that data on each network device. SNMP also defines a way for managed servers and devices (“agents”) to send event notification messages (“traps”) to management stations.

Before we delve into the arcana of SNMP, we should note that the terminology associated with it is some of the most wretched technobabble to be found in the networking arena. In many cases, the standard names for SNMP concepts and objects actively lead you away from an understanding of what’s going on.

That said, the protocol itself is simple; most of SNMP’s complexity lies above the protocol layer in the conventions for constructing the namespace and in the unnecessarily baroque vocabulary that surrounds SNMP like a protective shell. As long as you don’t think too hard about its internal mechanics, SNMP is easy to use.

SNMP was designed to be implementable by dedicated network hardware such as routers, in which context it remains a plausible option. SNMP was later extended to include monitoring of servers and desktop systems, but its fitness for this purpose has always been questionable. Today, much better alternatives (e.g., collectd; see this page) are available.

We suggest that you approach SNMP as a low-level data collection protocol for use with special-purpose devices that don’t support anything else. Get data out of the SNMP world as quickly as possible and turn it over to a general-purpose monitoring platform for storage and processing. SNMP can be an interesting neighborhood to visit, but you wouldn’t want to live there!
SNMP organization

SNMP data is arranged in a standardized hierarchy. The naming hierarchy is made up of “Management Information Bases” (MIBs), structured text files that describe the data accessible through SNMP. MIBs contain descriptions of specific data variables, which are referred to with names known as object identifiers, or OIDs. An OID is just a fancy way of naming a specific managed piece of information.

All current SNMP-capable devices support the structure for MIB-II defined in RFC1213. But each vendor can and does extend that MIB further to add more data and metrics.

OIDs exist within a hierarchical namespace where the nodes are numbered rather than named. However, for ease of reference, nodes also have conventional text names. The separator for pathname components is a dot. For example, the OID that refers to the uptime of a device is 1.3.6.1.2.1.1.3. This OID is also known by the human-readable (though not necessarily “human-understandable without additional documentation”) name

iso.org.dod.internet.mgmt.mib-2.system.sysUpTime

Table 28.3 presents a sampling of OID nodes that might be interesting to monitor for assessing network availability.

Table 28.3: Selected OIDs from MIB-II

<table>
<thead>
<tr>
<th>OID</th>
<th>Type</th>
<th>Contents</th>
</tr>
</thead>
<tbody>
<tr>
<td>system.sysDescr</td>
<td>string</td>
<td>System info: vendor, model, OS type, etc.</td>
</tr>
<tr>
<td>interfaces.ifNumber</td>
<td>int</td>
<td>Number of network interfaces present</td>
</tr>
<tr>
<td>interfaces.ifTable</td>
<td>table</td>
<td>Table of infobits about each interface</td>
</tr>
<tr>
<td>ip.ihForwarding</td>
<td>int</td>
<td>1 if system is a gateway; otherwise, 2</td>
</tr>
<tr>
<td>ip.ihAddrTable</td>
<td>table</td>
<td>Table of IP addressing data (masks, etc.)</td>
</tr>
<tr>
<td>icmp.icmPInRedirects</td>
<td>int</td>
<td>Number of ICMP redirects received</td>
</tr>
<tr>
<td>icmp.icmPInEchos</td>
<td>int</td>
<td>Number of pings received</td>
</tr>
<tr>
<td>tcp.tcppErrors</td>
<td>int</td>
<td>Number of TCP errors received</td>
</tr>
</tbody>
</table>


In addition to the universally supported MIB-II, there are MIBs for various kinds of hardware interfaces and protocols, MIBs for individual vendors, MIBs for different snmpd server implementations, and MIBs for particular hardware products.

A MIB is only a schema for naming management data. To be useful, a MIB must be backed up with agent-side code that maps between the SNMP namespace and the device’s actual state.

SNMP agents that run on UNIX, Linux, or Windows come with built-in support for MIB-II. Most can be extended to support supplemental MIBs and to interface with scripts that do the
actual work of fetching and storing these MIBs’ associated data. You’ll see lots of software like this that’s left over from a bygone era when SNMP was the new hotness. But it’s all smoke and no fire; you shouldn’t be running an SNMP agent on a UNIX system at all these days, except perhaps to answer the most basic queries about network configuration.
**SNMP protocol operations**

Only four basic SNMP operations exist: get, get-next, set, and trap.

Get and set are the basic operations for reading and writing data to the node identified by a specific OID. Get-next steps through a MIB hierarchy and can read the contents of tables as well.

A trap is an unsolicited, asynchronous notification from server (agent) to client (manager) that reports the occurrence of an interesting event or condition. Several standard traps are defined, including “I’ve just come up” notifications, reports of failure or recovery of a network link, and announcements of various routing and authentication problems. The mechanism by which the destinations of trap messages are specified depends on the implementation of the agent.

Since SNMP messages can potentially modify configuration information, some security mechanism is needed. The simplest version of SNMP security uses the concept of an SNMP “community string,” which is really just a horribly obfuscated way of saying “password.” There’s usually one community string for read-only access and another that allows writing.

Many systems come with the default community string set to “public”. Never leave this default in place; set real passwords for both the read-only and read/write community strings.

These days it makes a lot more sense to set up the SNMPv3 management framework, which allows for more security including authorization and access control for individual users.
Net-SNMP: tools for servers

On Linux and FreeBSD, the most common package that implements SNMP is called Net-SNMP. It includes an agent (snmpd), some command-line tools, a server for receiving traps, and even a library for developing SNMP-aware applications.

These days, Net-SNMP is primarily of interest because of its commands and libraries rather than its agent. It has been ported to many different UNIX-like systems, so it acts as a consistent platform on top of which you can write scripts. So, most distributions just separate out the Net-SNMP agent into a package of its own, making it easier to install only the commands.

On Debian and Ubuntu, the Net-SNMP packages are called snmp and snmpd. Install only the commands with `apt-get install snmp`.

On Red Hat and CentOS, the analogous packages are net-snmp and net-snmp-tools. Install the commands with `yum install net-snmp-tools`.

On Linux, configuration information goes in `/etc/snmp`; take note of the `snmpd.conf` file and snmp.d directory in that location. Run `systemctl start snmpd` to start up the agent daemon.

On FreeBSD, everything is included in one package: `pkg install net-snmp`. Configuration information goes in `/usr/local/etc/snmp`, which you’ll have to create by hand. You can start the agent by hand with `service snmpd start`, or put

```
snmpd_enable="YES"
```

in `/etc/rc.conf` to start it at boot time.

On all systems where you need to run the SNMP agent, you’ll need to ensure that UDP port 162 is not blocked by a firewall.

The commands that come with Net-SNMP can familiarize you with SNMP, and they’re also great for one-off checks of specific OIDs. Table 28.4 lists the most commonly used tools.

Table 28.4: Command-line tools in the Net-SNMP package
Basic SNMP checks generally use some combination of `snmpget` and `snmpdelta`. Other programs are helpful when you want to identify new OIDs to monitor from your fancy enterprise management tool. For example, `snmpwalk` starts at a specified OID (or at the beginning of the MIB, by default), and repeatedly makes “get next” calls to the agent. This process dumps a complete list of available OIDs and their associated values.

For example, here’s a truncated sample `snmpwalk` of the host tuva, a Linux system. The community string is “secret813community,” and `-v1` specifies simple authentication.

```
$ snmpwalk -c secret813community -v1 tuva
SNMPv2-MIB::sysDescr.0 = STRING: Linux tuva.atrust.com 2.6.9-11.Elsmp #1
SNMPv2-MIB::sysUpTime.0 = Timeticks: (1442) 0:00:14.42
SNMPv2-MIB::sysName.0 = STRING: tuva.atrust.com
IF-MIB::ifDescr.1 = STRING: lo
IF-MIB::ifDescr.2 = STRING: eth0
IF-MIB::ifDescr.3 = STRING: eth1
IF-MIB::ifType.1 = INTEGER: softwareLoopback(24)
IF-MIB::ifType.2 = INTEGER: ethernetCsmacd(6)
IF-MIB::ifType.3 = INTEGER: ethernetCsmacd(6)
IF-MIB::ifPhysAddress.1 = STRING:
IF-MIB::ifPhysAddress.2 = STRING: 0:11:43:d9:1e:f5
IF-MIB::ifPhysAddress.3 = STRING: 0:11:43:d9:1e:f6
IF-MIB::ifInOctets.1 = Counter32: 2605613514
IF-MIB::ifInOctets.2 = Counter32: 1543105654
IF-MIB::ifInOctets.3 = Counter32: 46312345
IF-MIB::ifInUcastPkts.1 = Counter32: 389536156
IF-MIB::ifInUcastPkts.2 = Counter32: 892959265
IF-MIB::ifInUcastPkts.3 = Counter32: 7712325
...
```

In this example, general information about the system is followed by statistics about the host’s network interfaces: lo0, eth0, and eth1. Depending on the MIBs supported by the agent you are managing, a complete dump can run to hundreds of lines. In fact, a full install on an Ubuntu system configured to serve every MIB spits out over 12,000 lines!
If you looked up the MIBs for the latest version of Net-SNMP on an Ubuntu system (check out mibdepot.com or install the **snmp-mibs-downloader** package), you would see that the five-minute load average OID is 1.3.6.1.4.1.2021.10.1.3.2. If you were interested in seeing the five-minute load average for the local host (configured with a community string of “public”), you would run:

```
$ snmpget -v 2c -c public localhost .1.3.6.1.4.1.2021.10.1.3.2
iso.3.6.1.4.1.2021.10.1.3.2 = STRING: "0.08"
```

Many useful SNMP-related Perl, Ruby, and Python modules are available from these languages’ respective module repositories. Although you can write scripts in terms of Net-SNMP commands, it’s usually easier and cleaner to make use of native modules that are customized for your language of choice.
28.10 Tips and Tricks for Monitoring

Over the years, we’ve picked up a few tips on how to maximize the effectiveness of monitoring. These are the main ideas:

- Avoid monitoring burn-out. Ensure that sysadmins who receive notifications outside of regular work hours get regular breaks. This goal is best achieved with a rotation system in which teams of two or more individuals are on call for a day or a week, then the next team takes over. Failure to heed this advice results in crabby sysadmins who hate their jobs.

- Define what circumstances really require 24 × 7 attention, and make sure this information is clearly communicated to your monitoring team, your on-call teams, and the customers or business units you support. The mere fact that you’re monitoring something doesn’t mean that administrators should be mustered at 3:30 a.m. when the value goes out of bounds. Many issues should be addressed during normal business hours.

- Eliminate monitoring noise. If false positives or notifications for noncritical services are being generated, make time to stop and fix them. Otherwise, like the boy who cried wolf, all notifications will eventually fail to receive the necessary attention.

- Create run books for everything. Any common restart, reset, or corrective procedure should be documented in a form that allows a responder who is not intimately familiar with the system in question to take appropriate action. The costs of not having such documentation are that problems will not be fixed quickly, mistakes will be made, and additional staff will be rousted to handle emergencies. Wikis are great for maintaining this type of documentation.

- Monitor the monitoring platform. This one will seem obvious once you’ve missed a critical outage because the monitoring platform was also down. Learn from our mistakes and make sure something is watching your watchful eyes.

- Miss an outage because of something that wasn’t monitored? Make sure it gets added so you catch the problem next time.

- Finally, and perhaps most importantly: no server or service goes into production without first being added to the monitoring system. No exceptions.
28.11 RECOMMENDED READING


Performance analysis and tuning are often treated as a form of system administration witchcraft. They’re not really witchcraft, but they do qualify as both science and art. The “science” part involves making careful quantitative measurements and applying the scientific method. The “art” part relates to the need to balance resources in a practical, level-headed way, since optimizing for one application or user can result in other applications or users suffering. As with so many things in life, you will often find that it’s impossible to make everyone happy.

Folks often assert that today’s performance problems are somehow wildly different from those of previous decades. That claim is inaccurate. It’s true that systems have become more complex, but the baseline determinants of performance and the high-level abstractions for measuring and managing it remain the same as always. Unfortunately, improvements in system performance correlate strongly with the community’s ability to create new applications that suck up all available resources.

An added complexity of recent years is the many layers of abstraction that often sit between your servers and the physical infrastructure of the cloud. It’s often impossible to know exactly what hardware is providing storage or CPU cycles to your server.

The magic and the challenge of the cloud are two aspects of the same form. Despite popular belief, you do not get to ignore performance considerations just because your servers are virtual. In fact, the billing models used by cloud providers create an even more direct link between
operational efficiency and server costs. Knowing how to measure and evaluate performance has become more important than ever.

This chapter focuses on the performance of systems that are used as servers. Desktop systems (and laptops) typically do not experience the same types of performance issues that servers do. The answer to the question of how to improve performance on a desktop machine is almost always, “Upgrade the hardware.” Users like this answer because it means they get fancy new systems more often.
29.1 PERFORMANCE TUNING PHILOSOPHY

One way that UNIX and Linux differ from other mainstream operating systems is that copious data are available to characterize their inner workings. Detailed information is generated by every level of the system, and administrators control a variety of tunable parameters. Source code is usually available for review if you have trouble identifying the cause of a performance problem despite the available instrumentation. For these reasons, UNIX and Linux are typically the operating systems of choice at performance-conscious sites.

Even so, performance tuning isn’t easy. Users and administrators alike often think that if they only knew the right “magic,” their systems would be twice as fast. But that’s rarely true.

One common fantasy involves tweaking the kernel variables that control the paging system and the buffer pools. These days, kernels are pretuned to achieve reasonable (though admittedly, not optimal) performance under a variety of load conditions. If you try to optimize the system on the basis of one particular measure of performance (e.g., buffer utilization), the chances are high that you will distort the system’s behavior relative to other performance metrics and load conditions.

The most serious performance issues often lie within applications and have little to do with the underlying operating system. This chapter discusses system-level performance tuning and mostly leaves application-level tuning to others. As a system administrator, you need to be mindful that application developers are people, too. How many times have you said, or thought, that “It must be a network problem”? In a similar way, application developers often initially assume that any issues must originate in a subsystem that is someone else’s responsibility.

Given the complexity of modern applications, some problems can only be resolved through collaboration among application developers, system administrators, server engineers, DBAs, storage administrators, and network architects. In this chapter, we help you determine what data and information to take back to these other folks to help them solve a performance problem—if, indeed, the problem lies in their area. This approach is far more productive than just saying, “Everything looks fine; it’s not my problem.”

In all cases, take everything you read on the Internet with a tablespoon of salt. In the area of system performance, you will see superficially convincing arguments on all sorts of topics. However, most of the proponents of these theories do not have the knowledge, discipline, and time required to design valid experiments. Popular support means very little; for every hare-brained proposal, you can expect to see a Greek chorus of “I increased the size of my buffer cache by a factor of ten just like Joe said, and the system feels much, much faster!!!” Right.

Here are some rules to keep in mind:

- Collect and review historical information about your system. If the system was performing fine a week ago, an examination of the aspects of the system that have changed may well lead you to a smoking gun. Keep baselines and trends in your hip
pocket to pull out in an emergency. As a first step, review log files to see if an underlying hardware problem has developed.

- Familiarize yourself with the trend collection and analysis tools discussed in Chapter 28, Monitoring. These tools are critical for performance assessment.

- Tune your system in a way that lets you compare the current results to the system’s previous baseline.

- Don’t intentionally overload your systems or your network. The kernel gives each process the illusion of infinite resources. But once 100% of the system’s resources are in use, the kernel has to work hard to maintain that illusion, delaying processes and often consuming a sizable fraction of the resources itself.

- As in particle physics, the more information you collect with system monitoring utilities, the more you affect the system you are observing. It is best to rely on something simple and lightweight that runs in the background (e.g., sar or vmstat) for routine observation. If those feelers show something significant, you can investigate further with other tools.

- When making changes, change only one thing at a time, and document each change that you make. Observe, record, and ponder the results before changing anything else.

- Always make sure you have a rollback plan in case your magic fix actually makes things worse.
29.2 WAYS TO IMPROVE PERFORMANCE

Here are some specific things you can do to improve performance:

- Ensure that the system has enough memory. As we see in the next section, memory size has a major influence on performance. If you’re running a system in the cloud, the amount of memory allocated to an instance is usually easy to adjust (though it’s often bundled with other resource allocations into a full-system profile).

- Eliminate storage resources’ dependence on mechanical operations where possible. Solid state disk drives (SSDs) are widely available and can yield big performance boosts because they don’t require the physical movement of a disk or armature to read bits. SSDs are easily installed (or, in the case of cloud environments, chosen) in place of existing old-school disk drives.

- If you are using UNIX or Linux as a web server or as some other type of network application server, you may want to spread traffic among several systems with a (physical or virtual) load balancer. Such an appliance balances the load according to one of several user-selectable algorithms such as “most responsive server” or “round robin.”

These load balancers also add useful redundancy should a server go down. They’re really quite necessary if your site must handle unexpected traffic spikes.

- Double-check the configuration of the system and of individual applications. Many applications can be tuned to yield tremendous performance improvements (e.g., by spreading data across disks, by not performing DNS lookups on the fly, or by running multiple instances of a server).

- Correct problems of usage, both those caused by “real work” (too many services running at once, inefficient programming practices, batch jobs run at excessive priority, and large jobs run at inappropriate times of day) and those caused by the system (such as unwanted daemons).

- Organize hard disks and filesystems so that load is evenly balanced, maximizing I/O throughput. For specific applications such as databases, you can use a fancy multidisk technology such as striped RAID to optimize data transfers. Consult your database vendor for recommendations. For Linux systems, ensure that you’ve selected the appropriate Linux I/O scheduler for your disk (see this page for details).

Remember that different types of applications and databases respond differently to being spread across multiple disks. RAID comes in many forms; take time to determine which form (if any) is appropriate for your particular application.
- Monitor your network to be sure that it is not saturated with traffic and that the error rate is low. A wealth of network information is available through the `netstat` (FreeBSD) and `ss` (Linux) commands.

- Identify cases where the system is fundamentally inadequate to satisfy the demands being made of it. You cannot tune your way out of these situations.

These steps are listed in rough order of effectiveness. Adding memory, converting to SSDs, and balancing traffic across multiple servers can often make a huge difference in performance. The effectiveness of the other measures ranges from noticeable to none.

Analysis and optimization of software data structures and algorithms almost always lead to significant performance gains. But unless you have a substantial base of local software, this level of design is usually out of your control.
29.3 Factors that Affect Performance

Perceived performance is determined by the basic capacities of the system’s resources and by the efficiency with which those resources are allocated and shared. The exact definition of a “resource” is rather vague. It can include such items as cached contexts on the CPU chip and entries in the address table of the memory controller. However, to a first approximation, only the following four resources have much effect on performance:

- CPU utilization (and stolen cycles, see below)
- Memory
- Storage I/O
- Network I/O

If resources are still left after active processes have taken what they want, the system’s performance is about as good as it can be.

If there are not enough resources to go around, processes must take turns. A process that does not have immediate access to the resources it needs has to wait around doing nothing. The amount of time spent waiting is one of the basic measures of performance degradation.

Historically, CPU utilization was one of the easiest resources to measure because a constant amount of processing power was always available. Nowadays, some virtualized or cloud environments may allocate CPU more dynamically. A process that’s using more than 90% of the allocated CPU is entirely CPU bound and is consuming essentially all of the system’s available computing power.

Many people assume that CPU resources are the most important factor affecting a system’s overall performance. Given infinite amounts of all other resources or certain types of applications (e.g., numerical simulations), a faster CPU (or more CPU cores) does make a dramatic difference. But in the everyday world, CPU is actually relatively unimportant.

Disk bandwidth is a common performance bottleneck. Because traditional hard disks are mechanical systems, it takes many milliseconds to locate a disk block, fetch its contents, and wake up the process that’s waiting for it. Delays of this magnitude overshadow every other source of performance degradation. Each disk access causes a stall worth millions of CPU instructions. Solid state drives (SSDs) are one tool you can use to address this problem.

Because of virtual memory, disk bandwidth and memory can be directly related if the demand for physical memory is greater than the supply. Situations in which physical memory becomes scarce often result in memory pages being written to disk so that they can be reclaimed and reused for another purpose. In these situations, using memory is just as expensive as using the disk. Avoid this trap when performance is important; make sure that every system has adequate physical memory.
Network bandwidth resembles disk bandwidth in many ways because of the latencies involved in network communication. However, networks are atypical in that they involve entire communities rather than individual computers. They are also particularly susceptible to hardware problems and overloaded servers.
29.4 Stolen CPU Cycles

The promise of the cloud (and of virtualization more generally) is that your server always has the resources it needs. In reality, much of this bounty is created with smoke and mirrors. Even in a large-scale virtualization environment, resource contention can have a noticeable effect on a virtual server’s performance.

CPU is the resource most commonly affected. There are two ways by which CPU cycles can be stolen from your virtual machine:

- The hypervisor running your VM enforces a CPU quota based on how much CPU power you have contracted for. Shortfalls can be addressed by allocation of more resources at the hypervisor level or by your purchasing a larger instance size from the cloud provider.

- The physical hardware is oversubscribed, and there are not enough physical CPU cycles available to meet the current needs of all VM instances, even though these instances may all be under their CPU quotas. On a cloud provider, fixing this issue may be as simple as restarting your instance so that it gets reassigned to new physical hardware. In a data center of your own, the solution may require upgrading your virtualization environment with more resources.

Although CPU stealing can happen to any operating system running on a virtualized platform, Linux gives you some visibility into this phenomenon with the `st` metric (“stolen”) in `top`, `vmstat`, and `mpstat`.

Here’s an example from `top`:

```
top - 18:36:42 up 3 days, 18:03, 1 user, load average: 3.40, 2.25, 2.08
Tasks: 218 total, 4 running, 217 sleeping, 0 stopped, 0 zombie
%Cpu: 41.6 us, 42.2 sy, 0.0 ni, 0.0 id, 0.0 wa, 0.0 hi, 0.0 si, 16.2 st
```

In this example, 16.2% of the time, the system is ready to do work but is unable to run because CPU is being diverted away from the VM by the hypervisor. This time spent waiting translates directly to reduced throughput. Monitor this metric carefully on virtual servers to ensure that your workloads aren’t being unintentionally starved of CPU.
29.5 Analysis of Performance Problems

It can be difficult to isolate performance problems in a complex system. As a sysadmin, you often receive anecdotal problem reports that suggest a particular cause or fix (e.g., “The web server has gotten painfully sluggish because of all those damn AJAX calls…”). Take note of this information, but don’t assume that it’s accurate or reliable; do your own investigation.

Rigorous, transparent application of the scientific method helps you reach conclusions that you and others in your organization can rely on. Such an approach lets others evaluate your results, increases your credibility, and raises the likelihood that your suggested changes will actually fix the problem.

“Being scientific” doesn’t mean that you have to gather all the relevant data yourself. External information usually helps a lot. Don’t spend hours doing experiments related to issues that can just as easily be looked up in a FAQ.

We suggest the following five steps:

1. Formulate the question. Pose a specific question in a defined functional area, or state a tentative conclusion or recommendation that you are considering. Be specific about the type of technology, the components involved, the alternatives you are considering, and the outcomes of interest.

2. Gather and classify evidence. Conduct a systematic search of documentation, knowledge bases, known issues, blogs, white papers, forums, and other resources to locate external evidence related to your question. On your own systems, capture telemetry data and, where necessary or possible, instrument specific system and application areas of interest.

3. Critically appraise the data. Review each data source for relevance and critique it for validity. Abstract key information and note the quality of the sources.

4. Summarize the evidence both narratively and graphically. Combine findings from multiple sources into a narrative précis and, if possible, a graphic representation. Data that appears equivocal in numeric form can often become decisive once charted.

5. Develop a conclusion statement. State your conclusions (i.e., the answer to your question) concisely. Assign a grade to indicate the overall strength or weakness of the evidence that supports your conclusions.
29.6 **SYSTEM PERFORMANCE CHECKUP**

Enough generalities—let’s look at some specific tools and areas of interest. Before you take measurements, you need to know what you’re looking at.
Taking stock of your equipment

Start your inquiry with an inventory of your (physical or virtual) hardware, especially CPU and memory resources. This inventory can help you interpret the information presented by other tools and can help you set realistic expectations regarding the upper bounds on performance.

On Linux systems, the /proc filesystem is the place to find an overview of the hardware your operating system thinks you have. (More detailed hardware information can be found in /sys; see this page.) Table 29.1 shows some of the key files. See this page for general information about /proc.

Table 29.1: Sources of hardware information on Linux

<table>
<thead>
<tr>
<th>File</th>
<th>Contents</th>
</tr>
</thead>
<tbody>
<tr>
<td>/proc/cpuinfo</td>
<td>CPU type and description</td>
</tr>
<tr>
<td>/proc/meminfo</td>
<td>Memory size and usage</td>
</tr>
<tr>
<td>/proc/diskstats</td>
<td>Disk devices and usage statistics</td>
</tr>
</tbody>
</table>

Four lines in /proc/cpuinfo help you identify the system’s exact CPU: vendor_id, cpu family, model, and model name. Some of the values are cryptic; it’s best to look up the decoding on-line.

The exact info contained in /proc/cpuinfo varies by system and processor, but here’s a representative example:

```bash
$ cat /proc/cpuinfo
processor : 0
vendor_id : GenuineIntel
cpu family : 6
model : 15
model name : Intel(R) Xeon(R) CPU E5310 @ 1.60GHz
stepping : 11
cpu MHz : 1600.003
cache size : 4096 KB
physical id : 0
cpu cores : 2
siblings : 2
...
```

The file contains one entry for each processor core seen by the OS. The data vary slightly by kernel version. The processor value uniquely identifies each core. physical id values are unique per CPU socket, and core id values (not shown above) are unique per core within a CPU socket. Cores that support hyperthreading (duplication of CPU contexts without duplication of other processing features) are identified by an ht in the flags field (not shown above). If hyperthreading is actually in use, the siblings field for each core shows how many contexts are available on a given core.
Another command to run for information on both FreeBSD and Linux is `dmidecode`. It dumps the system’s Desktop Management Interface (DMI, aka SMBIOS) data. The most useful option is `-t type`; Table 29.2 shows the valid types.

Table 29.2: Type values for `dmidecode -t`

<table>
<thead>
<tr>
<th>Value</th>
<th>Description</th>
</tr>
</thead>
<tbody>
<tr>
<td>1</td>
<td>System information</td>
</tr>
<tr>
<td>2</td>
<td>Base board Information</td>
</tr>
<tr>
<td>3</td>
<td>Chassis information</td>
</tr>
<tr>
<td>4</td>
<td>Processor information</td>
</tr>
<tr>
<td>7</td>
<td>Cache information</td>
</tr>
<tr>
<td>8</td>
<td>Port connector information</td>
</tr>
<tr>
<td>9</td>
<td>System slot information</td>
</tr>
<tr>
<td>11</td>
<td>OEM strings</td>
</tr>
<tr>
<td>12</td>
<td>System configuration options</td>
</tr>
<tr>
<td>13</td>
<td>BIOS language information</td>
</tr>
<tr>
<td>16</td>
<td>Physical memory array</td>
</tr>
<tr>
<td>17</td>
<td>Memory device</td>
</tr>
<tr>
<td>19</td>
<td>Memory array mapped address</td>
</tr>
<tr>
<td>32</td>
<td>System boot information</td>
</tr>
<tr>
<td>38</td>
<td>IPMI device information</td>
</tr>
</tbody>
</table>

The example below shows typical information:

```
$ sudo dmidecode -t 4
# dmidecode 2.11
SMBIOS 2.2 present.
Handle 0x0004, DMI type 4, 32 bytes.
Processor Information
   Socket Designation: PGA 370
   Type: Central Processor
   Family: Celeron
   Manufacturer: GenuineIntel
   ID: 65 06 00 00 FF F9 83 01
   Signature: Type 0, Family 6, Model 6, Stepping 5
...
```

Bits of network configuration information are scattered about the system. `ifconfig -a` (FreeBSD) and `ip a` (Linux) are the best sources of IP and MAC information for each configured interface.
Gathering performance data

Most performance analysis tools tell you what’s going on at a particular point. However, the number and character of loads probably change throughout the day. Be sure to gather a cross-section of data before taking action. The best information on system performance often becomes clear only after a long period (a month or more) of data collection. It is particularly important to collect data during periods of peak use. Resource limitations and system misconfigurations are often visible only when the machine is under heavy load. See Chapter 28, Monitoring, for more information about collecting and analyzing data.
Analyzing CPU usage

You will probably want to gather three kinds of CPU data: overall utilization, load averages, and per-process CPU consumption. Overall utilization can help identify systems on which the CPU’s speed is itself the bottleneck. Load averages profile overall system performance. Per-process CPU consumption data can identify specific processes that are hogging resources.

You can obtain summary information with the `vmstat` command. `vmstat` takes two arguments: the number of seconds to monitor the system for each line of output and the number of reports to print. If you don’t specify the number of reports, `vmstat` runs until you press <Control-C>. For example:

```
$ vmstat 5 5
procs ---------memory-------- -swap- ---io--- -system-- -----cpu-----
            r  b  swpd free  buff  cache  si  so  bi  bo  in  cs  us  sy  id  wa
 1 0  820 2666356 428776 487892  0  0 4741  65 1063 4857  25  1  73  0
 1 0  829 2570324 428812 519196  0  0 4613  11 1054 4732  25  1  74  0
 1 0  829 2539928 428852 535636  0  0 5099  13 1057 5219  90  1  9  0
 1 0  829 2472340 428920 581568  0  0 4536  10 1056 4686  87  3 10  0
 3 0  829 2449276 428960 605728  0  0 4818  21 1060 4943  20  3  77  0
```

Although exact columns vary among systems, CPU utilization statistics are fairly consistent across platforms. User time, system (kernel) time, idle time, and time waiting for I/O are shown in the `us`, `sy`, `id`, and `wa` columns on the far right. CPU numbers that are heavy on user time generally indicate computation, and high system numbers indicate that processes are making a lot of system calls or are performing lots of I/O.

A rule of thumb for general-purpose compute servers that has served us well over the years is this: the system should spend approximately 50% of its non-idle time in user space and 50% in system space; the overall idle percentage should be nonzero. If you are dedicating a server to a single, CPU-intensive application, the majority of time should be spent in user space.

The `cs` column shows context switches per interval (that is, the number of times that the kernel changed which process was running). The number of interrupts per interval (usually generated by hardware devices or components of the kernel) is shown in the `in` column. Extremely high `cs` or `in` values typically indicate a misbehaving or misconfigured hardware device. The other columns are useful for memory and disk analysis, which we discuss later in this chapter.

Long-term averages of the CPU statistics let you determine whether there is fundamentally enough CPU power to go around. If the CPU usually spends part of its time in the idle state, there are cycles to spare. Upgrading to a faster CPU won’t do much to improve the overall throughput of the system, although it may speed up individual operations.

As you can see from this example, the CPU generally flip-flops back and forth between heavy use and idleness. Therefore, be sure to observe these numbers as an average over time. The
smaller the monitoring interval, the less consistent the results.

On multiprocessor machines, most tools present an average of processor statistics across all processors. On Linux, the `mpstat` command generates `vmstat`-like output for each individual processor. The `-P` flag lets you specify a specific processor to report on. `mpstat` is useful for debugging software that supports symmetric multiprocessing—it’s also enlightening to see how (in)efficiently your system uses multiple processors. Here’s an example that shows the status of each of four processors:

```
linux$ mpstat -P ALL
08:13:38 PM CPU %user %nice %sys %iowait %irq %soft %idle intr/s
08:13:38 PM 0 1.02 0.00 0.49 1.29 0.04 0.38 96.79 473.93
08:13:38 PM 1 0.28 0.00 0.22 0.71 0.00 0.01 98.76 232.86
08:13:38 PM 2 0.42 0.00 0.36 1.32 0.00 0.05 97.84 293.85
08:13:38 PM 3 0.38 0.00 0.30 0.94 0.01 0.05 98.32 295.02
```

On a workstation with only one user, the CPU generally spends most of its time idle. Then when you render a web page or switch windows, the CPU is used heavily for a short period. In this situation, information about long-term average CPU usage is not meaningful.

The second CPU statistic that’s useful for characterizing the burden on your system is the “load average,” which represents the average number of runnable processes. It gives you a good idea of how many pieces the CPU pie is being divided into. The load average is obtained with the `uptime` command:

```
$ uptime
11:10am up 34 days, 18:42, 5 users, load average: 0.95, 0.38, 0.31
```

Three values are given, corresponding to the 1, 5, and 15-minute averages. In general, the higher the load average, the more important the system’s aggregate performance becomes. If there is only one runnable process, that process is usually bound by a single resource (commonly disk bandwidth or CPU). The peak demand for that one resource becomes the determining factor in performance.

When more processes share the system, loads may or may not be more evenly distributed. If the processes on the system all consume a mixture of CPU, disk, and memory, the performance of the system is less likely to be dominated by constraints on a single resource. In this situation, it becomes most important to look at average measures of consumption, such as total CPU utilization.

See this page for more information about priorities.

The system load average is an excellent metric to track as part of a system baseline. If you know your system’s load average on a normal day and it is in that same range on a bad day, this is a hint that you should look elsewhere (such as the network) for performance problems. A load
average above the expected norm suggests that you should look at the processes running on the system itself.

Another way to view CPU usage is to run the `ps -aux` command to see how much of the CPU each process is using. On a busy system, at least 70% of the CPU is often consumed by just one or two processes. Deferring the execution of the CPU hogs or reducing their priority makes the CPU more available to other processes.

See [this page](#) for more information about `top`.

An excellent alternative to `ps` is a program called `top`. It presents about the same information as `ps`, but in a live, regularly updated format that shows the status of the system over time. Refreshing `top`’s output too rapidly can itself be quite a CPU hog, so be judicious in your use of `top`. 
Understanding how the system manages memory

The kernel manages memory in units called pages that are usually 4KiB or larger. It allocates virtual pages to processes as they request memory. Each virtual page is mapped to real storage, either to RAM or to “backing store” on disk. The kernel uses a “page table” to keep track of the mappings between these made-up virtual pages and real pages of memory.

The kernel can effectively allocate as much memory as processes ask for by augmenting real RAM with swap space. Since processes expect their virtual pages to map to real memory, the kernel may have to constantly shuffle pages between RAM and swap as different pages are accessed. This activity is known as paging. Ages ago, a second process known as “swapping” could occur in which all of a process’s pages were pushed out to disk at the same time. Today, demand paging is used in all cases.

The kernel tries to manage the system’s memory so pages that have been recently accessed are kept in memory and less active pages are paged out to disk. This scheme is known as an LRU system since the least recently used pages are the ones that get shunted to disk.

It would be inefficient for the kernel to keep track of all memory references, so it uses a cache-like algorithm to decide which pages to keep in memory. The exact algorithm varies by system, but the concept is similar across platforms. This system is cheaper than a true LRU system and produces comparable results.

When memory is low, the kernel tries to guess which pages on the inactive list were least recently used. If those pages have been modified by a process, they are considered “dirty” and must be paged out to disk before the memory can be reused. Pages that have been laundered in this fashion (or that were never dirty to begin with) are “clean” and can be recycled for use elsewhere.

When a process refers to a page on the inactive list, the kernel returns the page’s memory mapping to the page table, resets the page’s age, and transfers the page from the inactive list to the active list. Pages that have been written to disk must be paged in before they can be reactivated if the page in memory has been remapped. A “soft fault” occurs when a process references an in-memory inactive page, and a “hard fault” results from a reference to a nonresident (paged-out) page. In other words, a hard fault requires a page to be read from disk and a soft fault does not.

The kernel tries to stay ahead of the system’s demand for memory, so there is not necessarily a one-to-one correspondence between page-out events and page allocations by running processes. The goal of the system is to keep enough free memory handy that processes don’t have to actually wait for a page-out each time they make a new allocation. If paging increases dramatically when the system is busy, it would probably benefit from more RAM.

You can tune the kernel’s “swappiness” parameter (/proc/sys/vm/swappiness) to tell the kernel how to balance between reclaiming swap-backed and file-backed pages. Setting
swappiness to zero focuses reclamations entirely on file-backed pages; a swappiness of 100 makes an equal balance between the two. By default, the swappiness parameter has a value of 60. (If you find yourself tempted to modify this parameter, it’s probably time to buy more RAM for the system.)

If the kernel is unable to reclaim pages, Linux uses an “out-of-memory killer” to handle this condition. The killer function selects and kills a process to free up memory. Although the kernel attempts to kill off the least important user process on your system, running out of memory is always something to avoid. In this situation, it’s likely that a substantial portion of the system’s resources are being devoted to memory housekeeping rather than to useful work.
Analyzing memory usage

Two numbers summarize memory activity: the total amount of active virtual memory and the current paging rate. The first number tells you the total demand for memory, and the second suggests the proportion of that memory that is actively used. Your goal is to reduce activity or increase memory until paging remains at an acceptable level. Occasional paging is inevitable; don’t try to eliminate it completely.

Run `swapon -s` to determine the amount of paging (swap) space that’s currently in use:

```
linux$ swapon -s
Filename Type Size Used Priority
/dev/hdb1 partition 4096532 0 -1
/dev/hda2 partition 4096564 0 -2
```

`swapon` reports usage in kilobytes. The sizes quoted by these programs do not include the contents of core memory, so you have to compute the total amount of virtual memory yourself:

$$VM = \text{size of real memory} + \text{amount of swap space used}$$

On FreeBSD systems, paging statistics obtained with `vmstat` look like this:

```
freebsd$ vmstat 5 5
procs memory page disks faults
r b w avm fre flt re pi po fr sr da0 cd0 in sy
0 0 0 412M 1.8G 97 0 1 0 200 6 0 0 51 359
2 0 0 412M 1.8G 1 0 0 0 0 7 0 0 5 27
2 0 0 412M 1.8G 0 0 0 0 0 7 0 0 4 25
1 0 0 412M 1.8G 0 0 0 0 0 6 0 0 4 25
0 0 0 412M 1.8G 0 0 0 0 0 7 0 0 6 26
```

CPU information has been removed from this example. Under the `procs` heading are shown the number of processes that are immediately runnable, blocked on I/O, and runnable but swapped. If the value in the `w` column is ever nonzero, it is likely that the system’s memory is pitifully inadequate relative to the current load.

Under the `memory` heading, you can see both the active virtual memory (`avm`) and free virtual memory (`fre`). The columns under the `page` heading give information about paging activity. All columns represent average values per second. Table 29.3 shows their meanings.

**Table 29.3: Decoding guide for FreeBSD vmstat paging statistics**
On Linux systems, paging statistics obtained with `vmstat` look like this:

```
linux$ vmstat 5 5
procs --------memory-------- -swap- ---io-- -system-- -----cpu-----
 r b spwd free buff cache si so bi bo in cs us sy id wa st
5 0 0 66488 46328 597972 0 0 252 45 1042 278 3 4 93 1 0
0 0 0 66364 46336 597972 0 0 0 37 1069 264 0 1 98 0 0
0 0 0 66364 46344 597972 0 0 0 5 1011 252 1 1 98 0 0
0 0 0 66364 46352 597972 0 0 0 3 1020 311 1 1 98 0 0
0 0 0 66364 46360 597972 0 0 0 21 1067 507 1 3 96 0 0
```

As in the FreeBSD output, the number of processes that are immediately runnable and that are blocked on I/O are shown under the `procs` heading. Paging statistics are condensed to two columns, `si` and `so`, which represent pages swapped in and out, respectively.

Any apparent inconsistencies among the memory-related columns are for the most part illusory. Some columns count pages and others count kilobytes. All values are rounded averages. Furthermore, some are averages of scalar quantities and others are average deltas.

Use the `si` and `so` fields to evaluate the system’s swapping behavior. A page-in (`si`) represents a page being recovered from the swap area. A page-out (`so`) represents data being written to the swap area after being forcibly ejected by the kernel.

If your system has a constant stream of page-outs, it’s likely that you would benefit from more physical memory. But if paging happens only occasionally and does not produce annoying hiccups or user complaints, you can ignore it. If your system falls somewhere in the middle, further analysis should depend on whether you are trying to optimize for interactive performance (e.g., a workstation) or for a more server-like workload.
Analyzing disk I/O

You can monitor disk performance with the `iostat` command. Like `vmstat`, it accepts optional arguments to specify an interval in seconds and a repetition count, and its first line of output is a summary since boot. `iostat` output on Linux looks like this:

```
linux$ iostat
... 
Device:  tps  kB_read/s  kB_wrtn/s  kB_read  kB_wrtn
sda  0.41    8.20     1.39   810865   137476
dm-0  0.39     7.87     1.27   778168   125776
dm-1  0.02     0.03     0.04    3220    3964
dm-2  0.01     0.23     0.06   22828    5652
```

Each hard disk has the columns `tps`, `kB_read/s`, `kB_wrtn/s`, `kB_read`, and `kB_wrtn`, indicating transfers per second, kilobytes read per second, kilobytes written per second, total kilobytes read, and total kilobytes written.

The cost of seeking is the most important factor affecting mechanical disk drive performance. To a first approximation, the rotational speed of the disk and the speed of the bus to which the disk is connected have relatively little impact. Modern mechanical disks can transfer hundreds of megabytes of data per second if they are read from contiguous sectors, but they can only perform about 100 to 300 seeks per second. If you transfer one sector per seek, you can easily realize less than 5% of the drive’s peak throughput. SSD disks have a significant advantage over their mechanical predecessors because their performance is not tied to platter rotation or head movement.

Whether you’re using mechanical or SSD disks, you should put filesystems that are used together on separate disks to maximize performance. Although bus architectures and device drivers influence efficiency, most computers can manage multiple disks independently, thereby increasing throughput. For example, it’s often worthwhile to segregate frequently accessed web server data and web server logs onto different disks.

It’s especially important to split the paging (swap) area among several disks if possible, since paging tends to slow down the entire system. Many systems can use both dedicated swap partitions and swap files on a formatted filesystem.

See this page for more information about `lsof` and `fuser`.

The `lsof` command, which lists open files, and the `fuser` command, which shows the processes that are using a filesystem, can be helpful for isolating disk I/O performance issues. These commands show interactions between processes and filesystems, some of which may be unintended. For example, if an application is writing its log to the same device used for database logs, a disk bottleneck may result.
fio: testing storage subsystem performance

fio ([github.com/axboe/fio](https://github.com/axboe/fio)) is available for both Linux and FreeBSD. Use it to test the performance of the storage subsystem. It’s particularly helpful in large environments where shared storage resources (such as a Storage Area Network) are deployed. If you find yourself in a situation where storage performance is a concern, it’s often valuable to determine quantitative values for the following:

- Throughput in I/O operations per second (IOPS) (read, write, and mixed)
- Average latency (read and write)
- Maximum latency (peak read or write latency)

As part of the fio distribution, config (.fio) files for common tests such as these are included in the examples subdirectory. Here’s an example of a simple read/write test:

```
$ fio read-write.fio
ReadWriteTest: (g=0): rw=rw, bs=4K-4K/4K-4K/4K-4K, eng=posixaio, depth=1
fio-2.18
Starting 1 thread
Jobs: 1 (f=1): [M] [100.0% done] [110.3MB/112.1MB/0KB /s]
[22.1K/20.4K/0 iops] [eta 00m:00s]
read: io=1024.7MB, bw=91326KB/s, iops=28801, runt= 9213msec
  slat (usec): min=0, max=73, avg= 2.30, stdev= 0.23
  clat (usec): min=0, max=2214, avg=20.30, stdev=101.20
  lat (usec): min=5, max=2116, avg=22.21, stdev=101.21
clat percentiles (usec):
| 1.00th=[  4],  5.00th=[  6], 10.00th=[  7], 20.00th=[  7],
| 30.00th=[  6], 40.00th=[  7], 50.00th=[  7], 60.00th=[  7],
| 70.00th=[  8], 80.00th=[  8], 90.00th=[  8], 95.00th=[ 10],
| 99.00th=[ 668], 99.50th=[ 1096], 99.90th=[ 1208], 99.95th=[ 1208],
| 99.99th=[ 1256]
...
READ: io=1024.7MB, aggrb=91326KB/s, minb=91326KB/s, maxb=91326KB/s,
    mint=10671msec, maxt=10671msec
WRITE: io=1023.4MB, aggrb=98202KB/s, minb=98202KB/s, maxb=98202KB/s,
    mint=10671msec, maxt=10671msec
```

As with so many performance-related metrics, there is no universally correct value for any of these measures. It’s best to establish a benchmark, make adjustments, and remeasure.
sar: collecting and reporting statistics over time

The sar command is a performance monitoring tool that has lingered through multiple UNIX and Linux epochs despite its somewhat obscure command-line syntax. The original command had its roots in early AT&T UNIX.

At first glance, sar seems to display much the same information as vmstat and iostat. However, there’s one important difference: sar can report on historical as well as current data.

The Linux package that contains sar is called sysstat.

Without options, the sar command reports CPU utilization for the day at 10-minute intervals since midnight, as shown below. This historical data collection is made possible by the sal script, which is part of the sar toolset and must be set up to run from cron at periodic intervals. sar stores the data it collects underneath the /var/log directory in a binary format.

```
linux$ sar
Linux 4.4.0-66-generic (nuerbull) 03/19/17 _x86_64_ (4 CPU)
19:10:01 CPU %user %nice %system %iowait %steal %idle
19:12:01 all 0.02 0.00 0.01 0.00 0.00 99.97
19:14:01 all 0.01 0.00 0.01 0.00 0.00 99.98
```

In addition to CPU information, sar can also report on metrics such as disk and network activity. Use sar -d for a summary of this day’s disk activity or sar -n DEV for network interface statistics. sar -A reports all available information.

See this page for more information about Grafana.

sar has some limitations, but it’s a good bet for quick-and-dirty historical information. If you’re serious about making a long-term commitment to performance monitoring, we suggest that you set up a data collection and graphing platform such as Grafana.
Choosing a Linux I/O scheduler

Linux systems use an I/O scheduling algorithm to mediate among processes competing to perform disk I/O. The I/O scheduler masses the order and timing of disk requests to achieve the best possible overall I/O performance for a given application or situation.

Three different scheduling algorithms are available in current Linux kernels:

- **Completely Fair Queuing**: This is the default algorithm and is usually the best choice for mechanical hard disks on general-purpose servers. It tries to evenly distribute access to I/O bandwidth. (If nothing else, the algorithm surely deserves an award for marketing: who could ever say no to a completely fair scheduler?)

- **Deadline**: This algorithm tries to minimize the latency for each request. It reorders requests to increase performance.

- **NOOP**: This algorithm implements a simple FIFO queue. It assumes that I/O requests have already been optimized or reordered by the driver, or that they will be optimized or reordered by the device (as might be done by an intelligent controller). This option may be the best choice in some SAN environments and is the best choice for SSD drives (because SSD drives don’t have variable retrieval latencies).

You can view or set the algorithm in use for any particular device through the file `/sys/block/disk/queue/scheduler`. The active scheduler is enclosed in brackets.

```bash
$ cat /sys/block/sda/queue/scheduler
noop deadline [cfq]
$ sudo sh -c "echo noop > /sys/block/sda/queue/scheduler"
$ cat /sys/block/sda/queue/scheduler
[noop] deadline cfq
```

By determining which scheduling algorithm is most appropriate for your environment (you may need to run trials with each scheduler) you may be able to improve I/O performance.

See [this page](#) for more information about GRUB.

Unfortunately, the scheduling algorithm does not persist across reboots when set in this manner. You can set it for all devices at boot time with the `elevator=algorithm` kernel argument. That configuration is usually set in the GRUB boot loader’s configuration file, `grub.conf`. 
perf: profiling Linux systems in detail

Linux kernel versions 2.6 and higher include a perf_events interface that affords user-level access to the kernel’s performance metric event stream. The perf command is a powerful, integrated system profiler that reads and analyzes information from this stream. All components of a system can be profiled: hardware, kernel modules, the kernel itself, shared libraries, and applications.

To get started with perf, you’ll need to get a full set of the linux-tools packages:

```bash
$ sudo apt-get install linux-tools-common linux-tools-generic
    linux-tools-'uname -r'
```

Once you’ve installed the software, check out the tutorial at goo.gl/f88mt for examples and use cases. (This is a deep link into perf.wiki.kernel.org.)

The TL;DR path to getting started is to try perf top, which is a top-like display of system-wide CPU use. Of course, the simple example below only scratches the surface of perf’s capabilities.

```bash
$ sudo perf top
```

Samples: 161K of event 'cpu-clock', Event count (approx.): 21695432426

<table>
<thead>
<tr>
<th>Overhead</th>
<th>Shared Object</th>
<th>Symbol</th>
</tr>
</thead>
<tbody>
<tr>
<td>4.63%</td>
<td>[kernel]</td>
<td>[k] 0x00007fff8183d3b5</td>
</tr>
<tr>
<td>2.15%</td>
<td>[kernel]</td>
<td>[k] finish_task_switch</td>
</tr>
<tr>
<td>2.04%</td>
<td>[kernel]</td>
<td>[k] entry_SYSCALL_64_after_swapgs</td>
</tr>
<tr>
<td>2.03%</td>
<td>[kernel]</td>
<td>[k] str2hashbuf_signed</td>
</tr>
<tr>
<td>2.00%</td>
<td>[kernel]</td>
<td>[k] half_md4_transform</td>
</tr>
<tr>
<td>1.44%</td>
<td>find</td>
<td>[.] 0x000000000000010a01</td>
</tr>
<tr>
<td>1.41%</td>
<td>[kernel]</td>
<td>[k] ext4_htree_store_dirent</td>
</tr>
<tr>
<td>1.21%</td>
<td>libc-2.23.so</td>
<td>[.] strlen</td>
</tr>
<tr>
<td>1.19%</td>
<td>[kernel]</td>
<td>[k] __d_lookup_rcu</td>
</tr>
<tr>
<td>1.14%</td>
<td>find</td>
<td>[.] 0x000000000000169f0</td>
</tr>
<tr>
<td>1.12%</td>
<td>[kernel]</td>
<td>[k] copy_user_generic_unrolled</td>
</tr>
<tr>
<td>1.06%</td>
<td>[kernel]</td>
<td>[k] kfree</td>
</tr>
<tr>
<td>1.06%</td>
<td>[kernel]</td>
<td>[k] __raw_spin_lock</td>
</tr>
<tr>
<td>1.03%</td>
<td>find</td>
<td>[.] 0x00000000000169fa</td>
</tr>
<tr>
<td>1.01%</td>
<td>find</td>
<td>[.] 0x0000000000016a05</td>
</tr>
<tr>
<td>0.86%</td>
<td>find</td>
<td>[.] fts_read</td>
</tr>
<tr>
<td>0.73%</td>
<td>[kernel]</td>
<td>[k] __kmalloc</td>
</tr>
<tr>
<td>0.71%</td>
<td>[kernel]</td>
<td>[k] ext4_readdir</td>
</tr>
<tr>
<td>0.69%</td>
<td>libc-2.23.so</td>
<td>[.] malloc</td>
</tr>
<tr>
<td>0.65%</td>
<td>libc-2.23.so</td>
<td>[.] fcntl</td>
</tr>
<tr>
<td>0.64%</td>
<td>[kernel]</td>
<td>[k] __ext4_check_dir_entry</td>
</tr>
</tbody>
</table>

The Overhead column shows the percentage of the time the CPU was in the corresponding function when it was sampled. The Shared Object column is the component (e.g., the kernel), shared library, or process in which the function resides, and the Symbol column is the name of the function (in cases where symbol information hasn’t been stripped).
29.7 Help! My server just got really slow!

In previous sections, we’ve talked mostly about issues that relate to the average performance of a system. Solutions to these long-term concerns generally take the form of configuration adjustments or upgrades.

However, you will find that even properly configured systems are sometimes more sluggish than usual. Luckily, transient problems are often easy to diagnose. Most of the time, they are caused by a greedy process that is simply consuming so much CPU power, disk, or network bandwidth that other processes are affected. On occasion, malicious processes hog available resources to intentionally slow a system or network, a scheme known as a “denial of service” (DOS) attack.

The first step in diagnosis is to run `ps auxww` or `top` to look for obvious runaway processes. Any process that’s using more than 50% of the CPU is likely to be at fault. If no single process is getting an inordinate share of the CPU, check to see how many processes are getting at least 10%. If you snag more than two or three (don’t count `ps` itself), the load average is likely to be quite high. This is, in itself, a cause of poor performance. Check the load average with `uptime`, and use `vmstat` or `top` to check whether the CPU is ever idle.

If no CPU contention is evident, run `vmstat` to see how much paging is going on. All disk activity is interesting: a lot of page-outs may indicate contention for memory, and disk traffic in the absence of paging may mean that a process is monopolizing the disk by constantly reading or writing files.

There’s no direct way to tie disk operations to processes, but `ps` can narrow down the possible suspects for you. Any process that is generating disk traffic must be using some amount of CPU time. You can usually make an educated guess about which of the active processes is the true culprit. Use `kill -STOP` to suspend the process and test your theory.

A large virtual address space or resident set used to be a suspicious sign, but shared libraries have made these numbers less useful. `ps` is not very smart about separating system-wide shared library overhead from the address spaces of individual processes. Many processes wrongly appear to have tens or hundreds of megabytes of active memory.

Suppose you do find that a particular process is at fault—what should you do? Usually, nothing. Some operations just require a lot of resources and are bound to slow down the system. It doesn’t necessarily mean that they’re illegitimate. It is sometimes useful, however, to `renice` an obtrusive process that is CPU-bound.

Sometimes, application tuning can dramatically reduce a program’s demand for CPU resources; this effect is especially visible with custom network server software such as web applications.

Processes that are disk or memory hogs often can’t be dealt with so easily. `renice` generally does not help. You do have the option of killing or stopping such processes, but we recommend...
against this if the situation does not constitute an emergency. As with CPU pigs, you can use the low-tech solution of asking the owner to run the process later.

Linux has a handy option for dealing with processes that consume excessive disk bandwidth in the form of the `ionice` command. This command sets a process’s I/O scheduling class; at least one of the available classes supports numeric I/O priorities (which can be set through `ionice` as well). The most useful invocation to remember is `ionice -c 3 -p pid`, which allows the named process to perform I/O only if no other processes wants to.

The kernel allows a process to restrict its own use of physical memory by calling the `setrlimit` system call. This facility is also available in the shell through the built-in `ulimit` command (limits on FreeBSD). For example, the command

```bash
$ ulimit -m 32000000
```

causes all subsequent commands that the user runs to have their use of physical memory limited to 32MB. This feature is roughly equivalent to `renice` for memory-bound processes. More granular resource management can be achieved through the Class-based Kernel Resource Management functionality; see ckrm.sourceforge.net.

If a runaway process doesn’t seem to be the source of poor performance, investigate two other possible causes. The first is an overloaded network. Many programs are so intimately bound up with the network that it’s hard to tell where system performance ends and network performance begins.

Some network overloading problems are hard to diagnose because they come and go very quickly. For example, if every machine on the network runs a network-related program out of `cron` at a particular time each day, there will often be a brief but dramatic glitch. Every machine on the net will hang for five seconds, and then the problem will disappear as quickly as it came.

Server-related delays are another possible cause of performance crises. UNIX and Linux systems are constantly consulting remote servers for NFS, Kerberos, DNS, and any of a dozen other facilities. If a server is dead or some other problem makes the server expensive to communicate with, the effects ripple back through client systems.

For example, on a busy system, some process may use the `gethostent` library routine every few seconds or so. If a DNS glitch makes this routine take two seconds to complete, you will likely perceive a difference in overall performance. DNS forward and reverse lookup configuration problems are responsible for a surprising number of server performance issues.
29.8 RECOMMENDED READING

DREPPER, ULRICH. *What Every Programmer Should Know about Memory.* You can find this article on-line at [lwn.net/Articles/250967](http://lwn.net/Articles/250967).


A service is only as reliable as the data center that houses it. For those with hands-on experience, that’s just common sense. (Of course, it’s possible to distribute a service among multiple data centers, thereby limiting the impact of a failure in any one center.)

Proponents of cloud computing sometimes seem to imply that the cloud can magically break the chains that join the physical and virtual worlds. But although cloud providers offer a variety of services that help boost resilience and availability, every cloud resource ultimately lives somewhere in mundane reality.

Understanding where your data actually lives is an important part of being a system administrator. If you are involved in selecting third party cloud providers, evaluate vendors and their facilities quantitatively. You might also find yourself in positions where security, data sovereignty, or political concerns dictate that you build and maintain your own data center.

A data center is composed of

- A physically safe and secure space
- Racks that hold computer, networking, and storage devices
- Electric power (and standby power) sufficient to operate the installed devices
- Cooling, to keep the devices within their operating temperature ranges
- Network connectivity throughout the data center, and to places beyond (enterprise network, partners, vendors, Internet)
• On-site operational staff to support the equipment and infrastructure

Certain aspects of data centers—such as their physical layout, power, and cooling—were traditionally designed and maintained by “facilities” or “physical plant” staff. However, the fast-moving pace of IT technology and the increasingly low tolerance for downtime have forced a shotgun marriage of IT and facilities staff as partners in the planning and operation of data centers.
30.1 Racks

The days of the traditional raised-floor data center—in which power, cooling, network connections, and telecommunications lines are all hidden underneath the floor—are over. Have you ever tried to trace a cable that runs under the floor of one of these labyrinths? Our experience is that although it looks nice through glass, a “classic” raised-floor machine room is really just a hidden rat’s nest. Today, you should use a raised floor to hide electrical power feeds, to distribute cooled air, and for nothing else. Network cabling (both copper and fiber) should be routed through overhead raceways designed specifically for this purpose. (These days, electrical feeds are often found overhead as well.)

In a dedicated data center, storing equipment in racks (as opposed to, say, setting it on tables or on the floor) is the only maintainable, professional choice. The best storage schemes use racks that are interconnected with an overhead track system through which cables can be routed. This approach confers that irresistible high-tech feel without sacrificing organization or maintainability.

The best overhead track system that we know of is manufactured by Chatsworth Products. You can construct homes for both shelf-mounted and rack-mounted servers from standard 19” single-rail telco racks. Two back-to-back 19” telco racks make a high-tech-looking “traditional” rack for situations in which you need to attach rack hardware both in front of and in back of equipment. Chatsworth provides the racks, cable races, and cable management doodads, as well as all the hardware necessary to mount them in your building. Since the cables lie in visible tracks, they are easy to trace and you will naturally be motivated to keep them tidy.
30.2 Power

Several strategies may be needed to provide a data center with clean, stable, fault-tolerant power. Common options include

- **Uninterruptible power supplies (UPSs)** – UPSs provide power when the normal long-term power source (e.g., the commercial power grid) becomes unavailable. Depending on size and capacity, a UPS can provide anywhere from a few minutes to a couple of hours of power. UPSs alone cannot support a site in the event of a long-term outage.

- **On-site power generation** – If the commercial grid is unavailable, on-site standby generators can provide long-term power. Generators are usually fueled by diesel, LP gas, or natural gas and can support the site as long as fuel is available. It is customary to store at least 72 hours of fuel on-site and to arrange to buy fuel from multiple providers.

- **Redundant power feeds** – In some locations, it may be possible to obtain more than one power feed from the commercial power grid (possibly from different power generators).

See [this page](#) for more information about shutdown procedures.

In all cases, servers and network infrastructure equipment should at least be put on uninterruptible power supplies. Good UPSs have an Ethernet or USB interface that can be attached either to the machine to which they supply power or to a centralized monitoring infrastructure that can elicit a higher-level response. Such connections let the UPS warn computers or operators that power has failed and that a clean shutdown should be performed before the batteries run out.

UPSs are available in various sizes and capacities, but even the largest ones cannot provide long-term backup power. If your facility must operate on standby power for longer than a UPS can handle, you need a local generator in addition to a UPS.

A large selection of standby power generators is available, ranging in capacity from 5 kW to more than 2,500 kW. The gold standard is the family of generators made by Cummins Onan (power.cummins.com). Most organizations select diesel as their fuel type. If you’re in a cold climate, make sure you fill the tank with “winter mix diesel” or substitute Jet A-1 aircraft fuel to prevent gelling. Diesel is chemically stable but can grow algae, so consider adding an algicide to diesel that you plan to store for an extended period.

Generators and the infrastructure to support them are expensive, but they can save money in some ways, too. If you install a standby generator, your UPSs need only be large enough to cover
the short gap between the power going out and your generator coming on-line.

If UPSs or generators are part of your power strategy, it is extremely important to have a periodic test plan in place. We recommend that you test all components of your standby power system at least every 6 months. In addition, you (or your vendor) should perform preventive maintenance on standby power components at least annually.
Rack power requirements

Planning the power for a data center is a difficult challenge. Typically, the opportunity to build a new data center (or to significantly remodel an existing one) comes up only every decade or so. So it’s important to look far down the road when planning power systems.

Most architects are biased toward calculating the amount of power needed in a data center by multiplying the center’s square footage by a magic number. This approach proves to be ineffective in most real-world cases because the size of the data center alone tells you little about the types of equipment it might eventually house. Our recommendation is to use a per-rack power consumption model and to ignore the amount of floor space.

Historically, data centers have been designed to provide between 1.5 kW and 3 kW to each rack. But now that server manufacturers have started squeezing servers into 1U of rack space and building blade server chassis that hold 20 or more blades, the power needed to support a full rack of modern gear has skyrocketed.

One approach to solving the power density problem is to put only a handful of 1U servers in each rack, leaving the rest of the rack empty. Although this technique eliminates the need to provide more power to the rack, it’s a prodigious waste of space. A better strategy is to develop a realistic projection of the power that might be needed by each rack and to provision power accordingly.

Equipment varies in its power requirements, and it’s hard to predict exactly what the future will hold. A good approach is to create a system of power consumption tiers that allocates the same amount of power to all racks in a particular tier. This scheme is useful not only for meeting current equipment needs but also for planning future use. Table 30.1 outlines some basic starting points for tier definitions.

Table 30.1: Power-tier estimates for racks in a data center

<table>
<thead>
<tr>
<th>Power tier</th>
<th>Watts/rack</th>
</tr>
</thead>
<tbody>
<tr>
<td>Insanely high density or “custom”</td>
<td>40 kW</td>
</tr>
<tr>
<td>Ultra high density</td>
<td>25 kW</td>
</tr>
<tr>
<td>Very high density (e.g., blade servers)</td>
<td>20 kW</td>
</tr>
<tr>
<td>High density (e.g., 1U servers)</td>
<td>16 kW</td>
</tr>
<tr>
<td>Storage equipment</td>
<td>12 kW</td>
</tr>
<tr>
<td>Network switching equipment</td>
<td>8 kW</td>
</tr>
<tr>
<td>Normal density</td>
<td>6 kW</td>
</tr>
</tbody>
</table>

Once you’ve defined your power tiers, estimate your need for racks in each tier. On the floor plan, put racks from the same tier together. Such zoning concentrates the high-power racks and lets you plan cooling resources accordingly.
kVA vs. kW

One of the many common disconnects between IT folks, facilities folks, and UPS engineers is that each of these groups uses different units for power. The amount of power a UPS can provide is typically labeled in kVA (kilovolt-amperes). But computer equipment and the electrical engineers that support your data center usually express power in watts (W) or kilowatts (kW). You might remember from fourth grade science class that watts = volts × amps. Unfortunately, your fourth grade teacher probably failed to mention that watts is a vector value, which for AC power includes a “power factor” (pf) in addition to volts and amps.

If you are designing a bottle-filling line at a brewery that involves lots of large motors and other heavy equipment, ignore this section and hire a qualified engineer to determine the correct power factor for use in your calculations. But for modern-day computer equipment, you can cheat and use a constant for a “probably good enough” conversion between kVA and kW:

\[ \text{kVA} = \frac{\text{kW}}{0.85} \]

See this page for some additional tips on measuring power consumption.

A final point to note is that when estimating the amount of power you need in a data center (or to size a UPS), you should measure devices’ actual power consumption rather than relying on manufacturers’ stated values as shown on equipment labels. Label values typically represent the maximum possible power consumption and are therefore misleading.
Energy efficiency

Energy efficiency has become a popular operational metric for evaluating data centers. The industry has standardized on a simple ratio known as the power usage effectiveness (PUE) as a way of expressing a plant’s overall efficiency:

\[
PUE = \frac{\text{Total power consumed by facility}}{\text{Total power consumed by IT equipment}}
\]

A hypothetically perfect data center would have a PUE of 1.0; that is, it would consume exactly the amount of power needed by IT gear, with no overhead. Of course, this goal is unreachable in practical terms. Equipment generates heat that must be removed, human operators need lighting and other environmental accommodations, etc. The higher the PUE value, the less energy efficient (and more expensive) a data center is to operate.

Modern-day data centers that are reasonably energy efficient generally have a PUE ratio of 1.4 or less. For reference, data centers from a decade ago typically had PUE ratios in the 2.0–3.0 range. Google, which has made energy efficiency a focus, regularly publishes its PUE ratios and as of 2016 has achieved an average PUE of 1.12 across its data centers.
**Metering**

You get what you measure. If you are serious about energy efficiency, it’s important to understand which devices are actually consuming the most energy. Although the PUE ratio gives you a general impression of the amount of energy consumed as non-IT overhead, it says very little about the power efficiency of the actual servers. (In fact, replacing servers with more power-efficient models will increase the PUE rather than decreasing it.)

It’s up to the data center administrator to select components that use the minimum amount of energy. One obvious enabling technology is power consumption metering at the aisle, rack, and device level. Select or build data centers that can easily provide this critical usage data.
Cost

Once upon a time, the cost of power was more or less the same across data centers in different locations. These days, the hyperscale cloud industry (Amazon, Google, Microsoft, and others) sends data center designers hunting for potential cost efficiencies in every corner of the world. One successful strategy has been to locate large data centers near sources of inexpensive power such as hydroelectric power plants.

When deciding whether to operate your own data center, be sure to factor the cost of power into your assessment. Chances are that the big guys have a built-in cost advantage in this aspect of operations (and others). Widespread fiber and bandwidth availability have largely rendered obsolete the traditional advice to locate your data center near your team.
Remote control

You might occasionally find yourself needing to regularly power-cycle a server because of a kernel or hardware glitch. (Or, perhaps you have non-Linux servers in your data center that are more prone to this type of problem.) In either case, you can consider installing a system that lets you power-cycle problem servers by remote control.

A reasonable family of solutions is manufactured by American Power Conversion (APC). Their remotely manageable products are conceptually similar to power strips, except that they can be controlled by a web browser that reaches the power distribution unit through a built-in Ethernet port.
30.3 Cooling and Environment

Just like humans, computers work better and live longer if they’re happy in their environment. Maintenance of a safe operating temperature is a prerequisite for this happiness.

The American Society of Heating, Refrigerating and Air-conditioning Engineers (ASHRAE) traditionally recommended data center temperatures (measured at server inlets) in the range of 68° to 77°F (20° to 25°C). To support organizations’ attempts to reduce energy consumption, ASHRAE released guidance in 2012 that suggests a more lenient temperature range to 64.4° to 80.6°F (18° to 27°C). Although this range seems unhelpfully broad, it does suggest that today’s hardware can flourish in a wide range of environments.
Cooling load estimation

Temperature maintenance starts with an accurate estimate of your cooling load. Traditional textbook models for data center cooling (even those from the 2000s) can be off from the realities of today’s high-density blade server chassis by up to an order of magnitude. Hence, we have found that it’s a good idea to double-check the cooling load estimates produced by your HVAC folks.

You need to determine the heat load contributed by the following components:

- Roof, walls, and windows
- Electronic gear
- Light fixtures
- Operators (people)

Of these, only the first should be left to your HVAC folks. The other components can be assessed by the HVAC team, but you should do your own calculations as well. Make sure that any discrepancies between your results and those of the HVAC team are fully explained before construction starts.

Roof, walls, and windows

Your roof, walls, and windows (don’t forget solar load) contribute to your environment’s cooling load. HVAC engineers usually have a lot of experience with these elements and should be able to give you good estimates.

Electronic gear

You can estimate the heat load produced by your servers (and other electronic gear) by determining their power consumption. In practical terms, all electric power that is consumed eventually ends up as heat.

As when planning for power-handling capacity, direct measurement of power consumption is by far the best way to obtain this information. Your friendly neighborhood electrician can help, or you can purchase an inexpensive meter and do it yourself. The Kill A Watt meter made by P3 is a popular choice at around $20, but it’s limited to small loads (15 amps) that plug in to a standard wall outlet. For larger loads or nonstandard connectors, use a clamp-on ammeter such as the Fluke 902 (also known as a “current clamp”) to make these measurements.

Most equipment is labeled with its maximum power consumption in watts. However, typical consumption tends to be significantly less than the maximum.

You can convert power consumption to the standard heat unit, BTUH (British thermal units per hour), by multiplying by 3.413 BTUH/watt. For example, if you wanted to build a data center that would house 25 servers rated at 450 watts each, the calculation would be
As with electronic gear, you can estimate light fixture heat load from power consumption. Typical office light fixtures contain four 40-watt fluorescent tubes. If your new data center had six of these fixtures, the calculation would be

\[
\left(6 \text{ fixtures}\right) \left(\frac{160 \text{ watts}}{\text{fixture}}\right) \left(3.412 \text{ BTU} / \text{watt}\right) = 3,276 \text{ BTU}
\]

Operators
At one time or another, humans will need to enter the data center to service something. Allow 300 BTUH for each occupant. To allow for four humans in the data center at the same time:

\[
\left(4 \text{ humans}\right) \left(\frac{300 \text{ BTU} }{\text{human}}\right) = 1,200 \text{ BTU}
\]

Total heat load
Once you have calculated the heat load for each component, sum the results to determine your total heat load. For this example, let’s assume that our HVAC engineer estimated the load from the roof, walls, and windows to be 20,000 BTUH.

20,000 BTUH for roof, walls, and windows
38,385 BTUH for servers and other electronic gear
3,276 BTUH for light fixtures
1,200 BTUH for operators

\[
\text{62,861 BTUH total}
\]

Cooling system capacity is typically expressed in tons. You can convert BTUH to tons by dividing by 12,000 BTUH/ton. You should also allow at least a 50% slop factor to account for errors and future growth.

\[
\left(62,681 \text{ BTUH}\right) \left(\frac{1 \text{ ton}}{12,000 \text{ BTUH}}\right) \left(1.5\right) = 7.86 \text{ tons of cooling required}
\]

See how your estimate matches up with the one from your HVAC folks.
**Hot aisles and cold aisles**

You can dramatically reduce your data center’s cooling difficulties by putting some thought into its physical layout. The most common and effective strategy is to alternate hot and cold aisles.

Facilities that have a raised floor and are cooled by a traditional CRAC (computer room air conditioner) unit are often set up so that cool air enters the space under the floor, rises up through holes in the perforated floor tiles, cools the equipment, and then rises to the top of the room as warm air, where it is sucked into return air ducts. Traditionally, racks and perforated tiles have been placed “randomly” about the data center, a configuration that results in relatively even temperature distribution. The result is an environment that is comfortable for humans but not really optimized for computers.

A better strategy is to lay out alternating hot and cold aisles between racks. Cold aisles have perforated cooling tiles and hot aisles do not. Racks are arranged so that equipment draws in air from a cold aisle and exhausts it to a hot aisle; the exhaust sides of two adjacent racks are therefore back to back. **Exhibit A** illustrates this basic concept.

**Exhibit A: Hot and cold aisles, raised floor**

![Hot and cold aisles, raised floor diagram]

This arrangement optimizes the flow of cooling so that air inlets always breathe cool air rather than another server’s hot exhaust. Properly implemented, the alternating row strategy results in aisles that are noticeably cold and hot. You can measure your cooling success with an infrared thermometer such as the Fluke 62, which is an indispensable tool of the modern system administrator. This point-and-shoot $100 device instantly measures the temperature of anything you aim it at, up to six feet away. Don’t take it out to the bars.

If you *must* run cabling under the floor, run power under cold aisles and network cabling under hot aisles.

Facilities without a raised floor can use in-row cooling units such as those manufactured by APC (apc.com). These units are skinny and sit between racks. **Exhibit B** shows how this system works.
Both CRAC and in-row cooling units need a way to dissipate heat outside the data center. This requirement is typically satisfied with a loop of liquid refrigerant (such as chilled water, Puron/R410A, or R22) that carries the heat outdoors. We omitted the refrigerant loops from Exhibit A and Exhibit B for simplicity, but most installations will require them.
**Humidity**

According to the 2012 ASHRAE guidelines, data center humidity should be kept between 8% and 60%. If the humidity is too low, static electricity becomes a problem. Recent testing has shown that there is little operational difference between 8% and the previous standard of 25%, so the minimum humidity standard was adjusted accordingly.

If humidity is too high, condensation can form on circuit boards and cause short circuits and oxidation.

Depending on your geographic location, you might need either humidification or dehumidification equipment to maintain a proper level of humidity.
Environmental monitoring

If you are supporting a mission-critical computing environment, it’s a good idea to monitor the temperature (and other environmental factors, such as noise and power) in the data center even when you are not there. It can be disappointing to arrive on Monday morning and find a pool of melted plastic on your data center floor.

Fortunately, automated data center monitors can watch the goods while you are away. We use and recommend the Sensaphone product family. These inexpensive boxes monitor environmental variables such as temperature, noise, and power, and they phone or text you when they detect a problem.
30.4 Data center reliability tiers

The Uptime Institute is a commercial entity that certifies data centers. They have developed a four-tier system for classifying the reliability of data centers, which we summarize in Table 30.2. In this table, N means that you have just enough of something (e.g., UPSs or generators) to meet normal needs. N+1 means that you have one spare; 2N means that each device has its own spare.

<table>
<thead>
<tr>
<th>Tier</th>
<th>Generators</th>
<th>UPSs</th>
<th>Power feeds</th>
<th>HVAC</th>
<th>Availability</th>
</tr>
</thead>
<tbody>
<tr>
<td>1</td>
<td>None</td>
<td>N</td>
<td>Single</td>
<td>N</td>
<td>99.671%</td>
</tr>
<tr>
<td>2</td>
<td>N</td>
<td>N+1</td>
<td>Single</td>
<td>N+1</td>
<td>99.741%</td>
</tr>
<tr>
<td>3</td>
<td>N+1</td>
<td>N+1</td>
<td>Dual, switchable</td>
<td>N+1</td>
<td>99.982%</td>
</tr>
<tr>
<td>4</td>
<td>2N</td>
<td>2N</td>
<td>Dual, simultaneous</td>
<td>2N</td>
<td>99.995%</td>
</tr>
</tbody>
</table>

a. With redundant components

Centers in the highest tier must be “compartmentalized,” which means that groups of systems are powered and cooled in such a way that the failure of one group has no effect on other groups.

Even 99.671% availability may look pretty good at first glance, but it works out to nearly 29 hours of downtime per year. 99.995% availability corresponds to 26 minutes of downtime per year.

Of course, no amount of redundant power or cooling will keep an application available if it’s administered poorly or is improperly architected. The data center is a foundational building block, necessary but not sufficient to ensure overall availability from the end user’s perspective.

You can learn more about the Uptime Institute’s certification standards (which include certification of design, construction, and operational phases) from their web site, uptimeinstitute.org. In some cases, organizations use the concept of these tiers without paying the Uptime Institute’s hefty certification fees. The important part is not the framed plaque but the use of a common vocabulary and assessment methodology to compare data centers.
30.5 **Data Center Security**

Perhaps it goes without saying, but the physical security of a data center is at least as important as its environmental attributes. Make sure that threats of both natural (e.g., fire, flood, earthquake) and human (e.g., competitors and criminals) origin have been carefully considered. A layered approach to security is the best way to ensure that a single mistake or lapse will not lead to a catastrophic outcome.
Location

Whenever possible, data centers should not be located in areas that are prone to forest fires, tornadoes, hurricanes, earthquakes, or floods. For similar reasons, it’s advisable to avoid man-made hazard zones such as airports, freeways, refineries, and tank farms.

Because the data center you select (or build) will likely be your home for a long time, it’s worthwhile to invest some time in researching the available risk data when making a site selection. The U.S. Geological Survey (usgs.gov) publishes statistics such as earthquake probability, and the Uptime Institute produces a composite map of data center location risks.
Perimeter

To reduce the risk of a targeted attack, a data center should be surrounded by a fence that is at least 25 feet from the building on all sides. Access to the inside of the fence perimeter should be controlled by a security guard or a multifactor badge access system. Vehicles allowed within the fence perimeter should not be permitted within 25 feet of the building.

Continuous video monitoring must cover 100% of the external perimeter, including all gates, access driveways, parking lots, and roofs.

Buildings should be unmarked. No signage should indicate what company the building belongs to or mention that it houses a data center.
Facility access

Access to the data center itself should be controlled by a security guard and a multifactor badge system, possibly one that incorporates biometric factors. Ideally, authorized parties should be enrolled in the physical access-control system before their first visit to the data center. If this is not possible, on-site security guards should follow a vetting process that includes confirming each individual’s identity and authorized actions.

One of the trickiest situations in training security guards is properly handling the appearance of “vendors” who claim they have come to fix some part of the infrastructure, such as the air conditioning. Make no mistake: unless the guard can confirm that someone authorized or requested this vendor visit, such visitors must be turned away.
Rack access

Large data centers are often shared with other parties. This is a cost-effective approach, but it comes with the added responsibility of securing each rack (or “cage of racks”). This is another case in which a multifactor access control system (such as a card reader plus a fingerprint reader) is needed to ensure that only authorized parties have access to your equipment. Each rack should also be individually monitored by video.
30.6 Tools

A well-outfitted sysadmin is an effective sysadmin. Having a dedicated tool box is an important key to minimizing downtime in an emergency. Table 30.3 lists some items to keep in your tool box, or at least within easy reach.

Table 30.3: A system administrator’s tool box

<table>
<thead>
<tr>
<th>General tools</th>
<th></th>
</tr>
</thead>
<tbody>
<tr>
<td>Hex (Allen) wrench kit</td>
<td>Ball-peen hammer, 4 oz.</td>
</tr>
<tr>
<td>Scissors</td>
<td>Electrician’s knife or Swiss Army knife</td>
</tr>
<tr>
<td>Small LED flashlight</td>
<td>Phillips-head screwdrivers: #0, #1, and #2</td>
</tr>
<tr>
<td>Socket wrench kit</td>
<td>Pliers, both flat-needlenose and regular</td>
</tr>
<tr>
<td>Stud finder</td>
<td>Ridgid SeeSnake micro inspection camera</td>
</tr>
<tr>
<td>Tape measure</td>
<td>Slot-head screwdrivers: 1/8&quot;, 3/16&quot;, and 5/16&quot;</td>
</tr>
<tr>
<td>Torx wrench kit</td>
<td>Teensy tiny jeweler’s screwdrivers</td>
</tr>
<tr>
<td>Tweezers</td>
<td></td>
</tr>
</tbody>
</table>

<table>
<thead>
<tr>
<th>Computer-related specialty items</th>
<th></th>
</tr>
</thead>
<tbody>
<tr>
<td>PC screw kit</td>
<td>Cable ties (and their Velcro cousins)</td>
</tr>
<tr>
<td>Infrared thermometer</td>
<td>Digital multimeter (DMM)</td>
</tr>
<tr>
<td>RJ-45 end crimper</td>
<td>Portable network analyzer/laptop</td>
</tr>
<tr>
<td>SCSI terminators</td>
<td>Spare Category 5 and 6A RJ-45 crossover cables</td>
</tr>
<tr>
<td>Spare power cord</td>
<td>Spare RJ-45 connectors (solid core and stranded)</td>
</tr>
<tr>
<td>Static grounding strap</td>
<td>Wire stripper (with an integrated wire cutter)</td>
</tr>
</tbody>
</table>

<table>
<thead>
<tr>
<th>Miscellaneous</th>
<th></th>
</tr>
</thead>
<tbody>
<tr>
<td>Q-Tips</td>
<td>Telescoping magnetic pickup wand</td>
</tr>
<tr>
<td>Cellular telephone</td>
<td>First-aid kit, including ibuprofen and acetaminophen</td>
</tr>
<tr>
<td>Electrical tape</td>
<td>Home phone and pager #s of on-call support staff</td>
</tr>
<tr>
<td>Can of compressed air</td>
<td>List of emergency maintenance contacts</td>
</tr>
<tr>
<td>Dentist’s mirror</td>
<td>Six-pack of good microbrew beer (suggested minimum)</td>
</tr>
</tbody>
</table>

a. And maintenance contract numbers, if applicable
30.7 **RECOMMENDED READING**


*Telecommunications Infrastructure Standard for Data Centers.* ANSI/TIA/EIA 942.

A variety of useful information and standards related to energy efficiency can be found at the Center of Expertise for Energy Efficiency in Data Centers web site at datacenters.lbl.gov.
During the past four decades, the role of information technology in business and daily life has changed dramatically. It’s hard to imagine a world without the instant gratification of Internet search.

For most of this period, the predominant philosophy of IT management was to increase stability by minimizing change. In many cases, hundreds or thousands of users depended on a single system. If a failure occurred, hardware often had to be express shipped for repair, or hours of downtime were needed to reinstall software and restore state. IT teams lived in fear that something would break and that they wouldn’t be able to fix it.

Change minimization has undesirable side effects. IT departments often became stuck in the past and failed to keep pace with business needs. “Technical debt” accumulated in the form of systems and applications in desperate need of upgrade or replacement that everyone was afraid to touch for fear of breaking something. IT staff became the butt of jokes and the least popular folks everywhere from board rooms to holiday parties.

Thankfully, those times are behind us. The advents of cloud infrastructure, virtualization, automation tools, and broadband communication have greatly reduced the need for one-off systems. Such servers have been replaced by armies of clones that are managed as battalions. In turn, these technical factors have enabled the evolution of a service philosophy known as DevOps which lets IT organizations drive and encourage change rather than resisting it. The DevOps name is a portmanteau of development and operations, the two traditional disciplines it combines.
An IT organization is more than a group of technical folks who set up Wi-Fi hot spots and computers. From a strategic perspective, IT is a collection of people and roles that use technology to accelerate and support the organization’s mission. Never forget the golden rule of system administration: enterprise needs drive IT activities, not the other way around.

In this chapter, we discuss the nontechnical aspects of running a successful IT organization that uses DevOps as its overarching schema. Most of the topics and ideas presented in this chapter are not specific to any particular environment. They apply equally to a part-time system administrator or to a large group of full-time professionals. Like green vegetables, they’re good for you no matter what size meal you’re preparing.
31.1 The Grand Unified Theory: DevOps

System administration and other operational roles within IT have traditionally been separate from domains such as application development and project management. The theory was that app developers were specialists who would push products forward with new features and enhancements. Meanwhile, the stolid and change-resistant operations team would provide 24 × 7 management of the production environment. This arrangement usually creates tremendous internal conflict and ultimately fails to meet the needs of the business and its clients.

Exhibit A: Courtesy of Dave Roth

The DevOps approach mingles developers (programmers, application analysts, application owners, project managers) with IT operations staff (system and network administrators, security monitors, data center staff, database administrators) in a tightly integrated way. This philosophy is rooted in the belief that working together as a collaborative team breaks down barriers, reduces finger pointing, and produces better results. Exhibit B summarizes a few of the main concepts.

Exhibit B: What is DevOps?
DevOps is a relatively new development in IT management. The early 2000s brought change to the development side of the house, which moved from “waterfall” release cycles to agile approaches that featured iterative development. This system increased the speed at which products, features, and fixes could be created, but deployment of those enhancements often stalled because the operations side wasn’t prepared to move as quickly as the development side. Hitching up the development and operations groups allowed everyone to accelerate down the road at the same pace, and DevOps was born.
**DevOps is CLAMS**

The tenets of DevOps philosophy are most easily described with the acronym CLAMS: Culture, Lean, Automation, Measurement, and Sharing.

**Culture**

People are the ultimate drivers of any successful team, so the cultural aspects of DevOps are the most important. Although DevOps has its own canon of cultural tips and tricks, the main goal is to get everyone working together and focused on the overall picture.

Under DevOps, all disciplines work together to support a common business driver (product, objective, community, etc.) through all phases of its life cycle. Achieving this goal may ultimately require changes in reporting structure (no more isolated application development groups), seating layout, and even job responsibilities. These days, good system administrators occasionally write code (often automation or deployment scripts), and good application developers regularly examine and manage infrastructure performance metrics.

Here are some typical features of a DevOps culture:

- Both developers (Dev) and operations (Ops) have 24 × 7, simultaneous (“everyone gets paged”), on-call responsibility for the complete environment. This rule has the wonderful side effect that root causes can be addressed wherever they occur. (The first six weeks or so of a shared on-call model is painful. Then suddenly it turns around. Trust us.)

- No application or service can launch without automated testing and monitoring being in place at both the system and application level. This rule seals in functionality and creates a contract between Dev and Ops. Likewise, Dev and Ops must sign off on any launch before it happens.

- All production environments are mirrored by identical development environments. This rule creates a runway for testing and reduces accidents in production.

- Dev teams do regular code reviews to which Ops is invited. Code architecture and functionality are no longer just Dev functions. Likewise, Ops performs regular infrastructure reviews in which Dev is involved. Dev must be aware of—and contribute to—decisions about underlying infrastructure.

- Dev and Ops have regular, joint stand-up meetings. In general, meetings should be minimized, but joint stand-ups serve as a useful stopgap to foster communication.

- Dev and Ops should all sit in a common chat room dedicated to discussion of both strategic (architecture, direction, sizing) and operational issues. This communication channel is often known as ChatOps, and several amazing platforms are available to support it. Check out HipChat, Slack, MatterMost, and Zulip, to name a few.
A successful DevOps culture pushes Dev and Ops so close that their scopes interpenetrate, and everyone learns to be comfortable with that. The optimal level of overlap is probably higher than most people would naturally prefer in the absence of cultural indoctrination. Team members must learn to respond gracefully to queries and feedback about their work from colleagues who may be formally trained in other disciplines.

Lean
The easiest way to explain the lean aspect of DevOps is to note that if you schedule a recurring weekly meeting at your organization to discuss your DevOps implementation plan, you have instantly failed.

DevOps is about real-time interaction and communication among people, processes, and systems. Use real-time tools (like ChatOps) to communicate wherever possible, and focus on solving component problems one at a time. Always ask “what can we do today” to make progress on an issue. Avoid the temptation to boil the ocean.

Automation
Automation is the most universally recognized aspect of DevOps. The two golden rules of automation are these:

- If you need to perform a task more than twice, it should be automated.
- Don’t automate what you don’t understand.

Automation brings many advantages:

- It prevents staff from being trapped performing mundane tasks. Staff brainpower and creativity can be used to solve new and more difficult challenges.
- It reduces the risk of human error.
- It captures infrastructure in the form of code, allowing versions and outcomes to be tracked.
- It facilitates evolution while also reducing risk. If a change fails, automated rollback is (well, should be) easy.
- It facilitates the use of virtualized or cloud resources to achieve scale and redundancy. Need more? Spin some up. Need less? Kill them off.

Tools are instrumental in the quest for automation. Systems such as Ansible, Salt, Puppet, and Chef (covered in Chapter 23, Configuration Management) are front and center. Continuous integration tools such as Jenkins and Bamboo (see this page) help manage repeatable or triggered tasks. Packaging and launch utilities such as Packer and Terraform automate low-level infrastructure tasks.
Depending on your environment, you might need one, some, or all (!?) of these tools. New tools and enhancements are being developed rapidly, so focus on finding the tool that is a good fit for the particular function or process you are automating, as opposed to picking a tool and then looking for the questions it answers. Most importantly, reevaluate your tool set every year or two.

Your automation strategy should include at least the following elements:

- **Automated setup of new machines:** This is not just OS installation. It also includes all the additional software and local configuration necessary to allow a machine to enter production. It’s inevitable that your site will need to support more than one type of configuration, so include multiple machine types in your plans from the beginning.

- **Automated configuration management:** Configuration changes should be entered in the configuration base and applied automatically to all machines of the same type. This rule helps keep the environment consistent.

- **Automated promotion of code:** Propagation of new functionality from the development environment to the test environment, and from the test environment into production, should be automated. Testing itself should be automated, with clear criteria for evaluation and promotion.

- **Systematic patching and updating of existing machines:** When you identify a problem with your setup, you need a standardized and easy way to deploy updates to all affected machines. Because servers are not turned on all the time (even if they are supposed to be), your update scheme must correctly handle machines that are not online when an update is initiated. You can check for updates at boot time or update on a regular schedule; see *Periodic processes* for more information.

**Measurement**

See *Chapter 28* for more information about monitoring.

The ability to scale virtualized or cloud infrastructure (see *Chapter 9, Cloud Computing*) has pushed the world of instrumentation and measurement to new heights. Today’s gold standard is the collection of sub-second measurements throughout the entire service stack (business, application, database, subsystems, servers, network, and so on). Several DevOps-y tools such as Graphite, Grafana, ELK (the Elasticsearch + Logstash + Kibana stack), plus monitoring platforms like Icinga and Zenoss, support these efforts.

Having measurement data and doing something useful with it are two different things, however. A mature DevOps shop ensures that metrics from the environment are visible and evangelized to all interested parties (both inside and outside of IT). DevOps sets nominal targets for each metric and chases down any anomalies to determine their cause.
Sharing

Collaborative work and shared development of capabilities lie at the heart of a successful DevOps effort. Staff should be encouraged and incentivized to share their work both internally (lunch-and-learn presentations, team show-and-tell, wiki how-to articles) and externally (Meetups, white papers, conferences). These efforts extend the silo-busting philosophy beyond the local workgroup and help everyone learn and grow.
System administration in a DevOps world

System administrators have always been the jacks-and-jills-of-all-trades of the IT world, and that remains true under the broader DevOps umbrella. The system administrator role oversees systems and infrastructure, typically including primary responsibility for these areas:

- Building, configuring, automating, and deploying system infrastructure
- Ensuring that the operating system and major subsystems are secure, patched, and up to date
- Deploying, supporting, and evangelizing DevOps technologies for continuous integration, continuous deployment, monitoring, measurement, containerization, virtualization, and ChatOps platforms
- Coaching other team members on infrastructure and security best practices
- Monitoring and maintaining infrastructure (physical, virtual, or cloud) to ensure that it meets performance and availability requirements
- Responding to user resource or enhancement requests
- Fixing problems with systems and infrastructure as they occur
- Planning for the future expansion of systems, infrastructure, and capacity
- Advocating cooperative interactions among team members
- Managing various outside vendors (cloud, co-location, disaster recovery, data retention, connectivity, physical plant, hardware service)
- Managing the life cycle of infrastructure components
- Maintaining an emergency stash of ibuprofen, tequila, and/or chocolate to be shared with other team members on those not-as-fresh days

This is just a subset of the breadth covered by a successful system administrator. The role is part drill sergeant, part mother hen, part EMT, and part glue that keeps everything running smoothly.

Above all, remember that DevOps is founded on overcoming one’s normal territorial impulses. If you find yourself at war with other team members, take a step back and remember that you are most effective if you are seen as a hero who helps make everyone else successful.
31.2 Ticketing and Task Management Systems

A ticketing and task management system lies at the heart of every functioning IT group. As with all things DevOps, having one ticketing system that spans all IT disciplines is critical. In particular, enhancement requests, issue management, and software bug tracking should all be part of the same system.

A good ticketing system helps staff avoid two of the most common workflow pitfalls:

- Tasks that fall through the cracks because everyone thinks they are being taken care of by someone else
- Resources that are wasted through duplication of effort when multiple people or groups work on the same problem without coordination
Common functions of ticketing systems

A ticket system accepts requests through various interfaces (email and web being the most common) and tracks them from submission to solution. Managers can assign tickets to groups or to individual staff members. Staff can query the system to see the queue of pending tickets and perhaps resolve some of them. Users can find out the status of requests and see who is working on them. Managers can extract high-level information such as

- The number of open tickets
- The average time to close a ticket
- The productivity of staff members
- The percentage of unresolved (rotting) tickets
- Workload distribution by time to solution

The request history stored in the ticket system becomes a history of the problems with your IT infrastructure and also the solutions to those problems. If that history is easily searchable, it becomes an invaluable resource for the sysadmin staff.

Resolved tickets can be provided to novice staff members and trainees, inserted into a FAQ system, or made searchable for later discovery. New staff members can benefit from receiving copies of closed tickets because those tickets include not only technical information but also examples of the tone and communication style that are appropriate for use with customers.

Like all documents, your ticketing system’s historical data can potentially be used against your organization in court. Follow the document retention guidelines set up by your legal department.

Most request tracking systems automatically confirm new requests and assign them a tracking number that submitters can use to follow up or inquire about the request’s status. The automated response message should clearly state that it is just a confirmation. It should be followed promptly by a message from a real person that explains the plan for dealing with the problem or request.
Ticket ownership

Work can be shared, but in our experience, responsibility is less amenable to distribution. Every task should have a single, well-defined owner. That person need not be a supervisor or manager, just someone willing to act as a coordinator—someone willing to say, “I take responsibility for making sure this task gets done.”

An important side effect of this approach is that it is implicitly clear who implemented what or who made what changes. This transparency becomes important if you want to figure out why something was done in a certain way or why something is suddenly working differently or not working anymore.

Being “responsible” for a task should not equate to being a scapegoat if problems arise. If your organization defines responsibility as blameworthiness, you may find that the number of available project owners quickly dwindles. Your goal in assigning ownership is simply to remove ambiguity about who should be addressing each problem. Don’t punish staff members for requesting help.

From a customer’s point of view, a good assignment system is one that routes problems to a person who is knowledgeable and can solve them quickly and completely. But from a managerial perspective, assignments must occasionally be challenging to the assignee so that staff members continue to grow and learn in the course of doing their jobs. Your task is to balance reliance on staff members’ strengths against the need to challenge them, all while keeping both customers and staff members happy.

Larger tasks can be anything up to and including full-blown software engineering projects. These tasks may require the use of formal project management and software engineering tools. We don’t describe those tools here; nevertheless, they’re important and should not be overlooked.

Sometimes sysadmins know that a particular task needs to be done, but they don’t do it because the task is unpleasant. A sysadmin who points out a neglected, unassigned, or unpopular task is likely to receive that task as an assignment. This situation creates a conflict of interest because it motivates sysadmins to remain silent regarding such situations. Don’t let that happen at your site; give your sysadmins an avenue for pointing out problems. You can allow them to open up tickets without assigning an owner or associating themselves to the issue, or you can create an email alias to which issues can be sent.
User acceptance of ticketing systems

Receiving a prompt response from a real person is a critical determinant of customer satisfaction, even if the personal response contains no more information than the automated response. For most problems, it is far more important to let the submitter know that the ticket has been reviewed by a real person than it is to fix the problem immediately. Users understand that administrators receive many requests, and they’re willing to wait a fair and reasonable time for your attention. But they’re not willing to be ignored.

The mechanism through which users submit tickets affects their perception of the system. Make sure you understand your organization’s culture and your users’ preferences. Do they want a web interface? A custom application? An email alias? Maybe they’re only willing to make phone calls!

It’s also important that administrators take the time to make sure they understand what users are actually requesting. This point sounds obvious, but think back to the last five times you emailed a customer service or tech support alias. We’d bet there were at least a couple of cases in which the response seemed to have nothing to do with the question—not because those companies were especially incompetent, but because accurately parsing tickets is harder than it looks.

Once you’ve read enough of a ticket to develop an impression of what the customer is asking about, the rest of the ticket starts to look like “blah blah blah.” Fight this! Clients hate waiting for a ticket to find its way to a human, only to learn that the request has been misinterpreted and must be resubmitted or restated. Back to square one.

Tickets are often vague or inaccurate because the submitter does not have the technical background needed to describe the problem in the way that a sysadmin would. That doesn’t stop users from making their own guesses as to what’s wrong, however. Sometimes these guesses are perfectly correct. Other times, you must first decode the ticket to determine what the user thinks the problem is, then trace back along the user’s train of thought to intuit the underlying problem.
Sample ticketing systems

The following tables summarize the characteristics of several well-known ticketing systems. Table 31.1 shows open source systems, and Table 31.2 shows commercial systems.

Table 31.1: Open source ticket systems

<table>
<thead>
<tr>
<th>Name</th>
<th>Input</th>
<th>Lang</th>
<th>Back end</th>
<th>Web site</th>
</tr>
</thead>
<tbody>
<tr>
<td>Double Choco Latte</td>
<td>W</td>
<td>PHP</td>
<td>MP</td>
<td>github.com/gnuedcl/dcl</td>
</tr>
<tr>
<td>Mantis</td>
<td>WE</td>
<td>PHP</td>
<td>M</td>
<td>mantisbt.org</td>
</tr>
<tr>
<td>OTRS</td>
<td>WE</td>
<td>Perl</td>
<td>DMOP</td>
<td>otrs.org</td>
</tr>
<tr>
<td>RT: Request Tracker</td>
<td>WE</td>
<td>Perl</td>
<td>M</td>
<td>bestpractical.com</td>
</tr>
<tr>
<td>OSTicket</td>
<td>WE</td>
<td>PHP</td>
<td>M</td>
<td>osticket.com</td>
</tr>
<tr>
<td>Bugzilla</td>
<td>WE</td>
<td>Perl</td>
<td>MOP</td>
<td>bugzilla.org</td>
</tr>
</tbody>
</table>

a. Input types: W = web, E = email
b. Back end: D = DB2, M = MySQL, O = Oracle, P = PostgreSQL

Table 31.2 shows some of the commercial alternatives for request management. Since the web sites for commercial offerings are mostly marketing hype, details such as the implementation language and back end are not listed.

Table 31.2: Commercial ticket systems

<table>
<thead>
<tr>
<th>Name</th>
<th>Scale</th>
<th>Web site</th>
</tr>
</thead>
<tbody>
<tr>
<td>EMC Ionix (Infra)</td>
<td>Huge</td>
<td>infra-corp.com/solutions</td>
</tr>
<tr>
<td>HEAT</td>
<td>Medium</td>
<td>ticomix.com</td>
</tr>
<tr>
<td>Jira</td>
<td>Any</td>
<td>atlassian.com</td>
</tr>
<tr>
<td>Remedy (now BMC)</td>
<td>Huge</td>
<td>remedy.com</td>
</tr>
<tr>
<td>ServiceDesk</td>
<td>Huge</td>
<td>ca.com/us/service-desk.aspx</td>
</tr>
<tr>
<td>ServiceNow</td>
<td>Any</td>
<td>servicenow.com</td>
</tr>
<tr>
<td>Track-It!</td>
<td>Medium</td>
<td>trackit.com</td>
</tr>
</tbody>
</table>

Some of the commercial offerings are so complex that they need a person or two dedicated to maintaining, configuring, and keeping them running. Others (such as Jira and ServiceNow) are available as a “software as a service” product.
Ticket dispatching

In a large group, even one with an awesome ticketing system, one problem still remains to be solved: it is inefficient for several people to divide their attention between the task they are working on right now and the request queue, especially if requests come in by email to a personal mailbox. We have experimented with two solutions to this problem.

Our first try was to assign half-day shifts of trouble queue duty to staff members in our sysadmin group. The person on duty would try to answer as many of the incoming queries as possible during a shift. The problem with this approach was that not everybody had the skills to answer all questions and fix all problems. Answers were sometimes inappropriate because the person on duty was new and was not familiar with the customers, their environments, or the specific support contracts they were covered by. The result was that the more senior people had to keep an eye on things and so were not able to concentrate on their own work. In the end, the quality of service was worse and nothing was really gained.

After this experience, we created a “dispatcher” role that rotates monthly among a group of senior administrators. The dispatcher checks the ticketing system for new entries and farms out tasks to specific staff members. If necessary, the dispatcher contacts users to extract any additional information that is necessary for prioritizing requests. The dispatcher uses a home-grown database of staff skills to decide who on the support team has the appropriate skills and time to address a given ticket. The dispatcher also makes sure that requests are resolved in a timely manner.
31.3 Local documentation maintenance

Just as most people accept the health benefits of exercise and leafy green vegetables, everyone appreciates good documentation and has a vague idea that it’s important. Unfortunately, that doesn’t necessarily mean that they’ll write or update documentation without prodding. Why should we care, really?

- Documentation reduces the likelihood of a single point of failure. It’s wonderful to have tools that deploy workstations in no time and distribute patches with a single command, but these tools are nearly worthless if no documentation exists and the expert is on vacation or has quit.

- Documentation aids reproducibility. When practices and procedures are not stored in institutional memory, they are unlikely to be followed consistently. When administrators can’t find information about how to do something, they have to wing it.

- Documentation saves time. It doesn’t feel like you’re saving time as you write it, but after spending a few days re-solving a problem that has been tackled before but whose solution has been forgotten, most administrators are convinced that the time is well spent.

- Finally, and most importantly, documentation enhances the intelligibility of the system and allows subsequent modifications to be made in a manner that’s consistent with the way the system is supposed to work. When modifications are made on the basis of only partial understanding, they often don’t quite conform to the architecture. Entropy increases over time, and even the administrators that work on the system come to see it as a disorderly collection of hacks. The end result is often the desire to scrap everything and start again from scratch.

Local documentation should be kept in a well-defined spot such as an internal wiki or a third-party service such as Google Drive. Once you have convinced your administrators to document configurations and administration practices, it’s important to protect this documentation as well. A malicious user can do a lot of damage by tampering with your organization’s documentation. Make sure that people who need the documentation can find it and read it (make it searchable), and that everyone who maintains the documentation can change it. But balance accessibility with the need for protection.
**Infrastructure as code**

Another important form of documentation is known as “infrastructure as code.” It can take a variety of forms, but is most commonly seen in the form of configuration definitions (such as Puppet modules or Ansible playbooks) that can then be stored and tracked in a version control system such as Git. The system and its changes are well documented in the configuration files, and the environment can be built and compared against the standard on a regular basis. This approach ensures that the documentation and the environment always match and are up to date, solving the most common problem of traditional documentation. See Chapter 23, *Configuration Management*, for more information.
**Documentation standards**

If you must document elements manually, our experience suggests that the easiest and most effective way to maintain documentation is to standardize on short, lightweight documents. Instead of writing a system management handbook for your organization, write many one-page documents, each of which covers a single topic. Start with the big picture and then break it down into pieces that contain additional information. If you have to go into more detail somewhere, write an additional one-page document that focuses on steps that are particularly difficult or complicated.

This approach has several advantages:

- Higher management is probably only interested in the general setup of your environment. That is all that’s needed to answer questions from above or to conduct a managerial discussion. Don’t pour on too many details or you will just tempt your boss to intervene in them.

- The same holds true for customers.

- A new employee or someone taking on new duties within your organization needs an overview of the infrastructure to become productive. It’s not helpful to bury such people in information.

- It’s more efficient to use the right document than to browse through a large document.

- You can index pages to make them easy to find. The less time administrators have to spend looking for information, the better.

- It’s easier to keep documentation current when you can do that by updating a single page.

This last point is particularly important. Keeping documentation up to date is a huge challenge; documentation is often the first thing to be dropped when time is short. We have found that a couple of specific approaches keep the documentation flowing.

First, set the expectation that documentation be concise, relevant, and unpolished. Cut to the chase; the important thing is to get the information down. Nothing makes the documentation sphincter snap shut faster than the prospect of writing a dissertation on design theory. Ask for too much documentation and you might not get any. Consider developing a simple form or template for your sysadmins to use. A standard structure helps avoid blank-page anxiety and guides sysadmins to record pertinent information rather than fluff.

Second, integrate documentation into processes. Comments in configuration files are some of the best documentation of all. They’re always right where you need them, and maintaining them takes virtually no time at all. Most standard configuration files allow comments, and even those
that aren’t particularly comment friendly can often have some extra information sneaked into them.

Locally built tools can require documentation as part of their standard configuration information. For example, a tool that sets up a new computer can require information about the computer’s owner, location, support status, and billing information, even if these facts aren’t directly relevant to the machine’s software configuration.

See this page for more information about cron.

Documentation should not create information redundancies. For example, if you maintain a site-wide master list of systems, there should be no other place where this information is updated by hand. Not only is it a waste of your time to make updates in multiple locations, but inconsistencies are also certain to creep in over time. When this information is required in other contexts and configuration files, write a script that obtains it from (or updates) the master configuration. If you cannot completely eliminate redundancies, at least be clear about which source is authoritative. And write tools to catch inconsistencies, perhaps run regularly from cron.

The advent of tools such as wikis, blogs, and other simple knowledge management systems has made it much easier to keep track of IT documentation. Set up a single location where all your documents can be found and updated. Don’t forget to keep it organized, however. One wiki page with 200 child pages all in one list is cumbersome and difficult to use. Be sure to include a search function to get the most out of your system.
31.4 Environment separation

Organizations that write and deploy their own software need separate development, test, and production environments so that releases can be staged into general use through a structured process. Separate, that is, but identical; make sure that when development systems are updated, the changes propagate to the test and production environments as well. Of course, the configuration updates themselves should be subject to the same kind of structured release control as the code. “Configuration changes” include everything from OS patches to application updates and administrative changes.

Historically, it has been standard practice to “protect” the production environment by enforcing role separation throughout the promotion process. For example, the developers who have administrative privileges in the development environment are not the same people who have administrative and promotion privileges in other environments. The fear was that a disgruntled developer with code promotion permissions could conceivably insert malicious code at the development stage and then promote it through to production. By distributing approval and promotion duties to other people, multiple people would need to collude or make mistakes before problems could find their way into production systems.

Unfortunately, the anticipated benefits of such draconian measures are rarely realized. Code promoters often don’t have the skills or time to review code changes at a level that would actually catch intentional mischief. Instead of helping, the system creates a false sense of protection, introduces unnecessary roadblocks, and wastes resources.

In the DevOps era, this problem is solved in a different way. Rather than separate roles, the preferred approach is to track all changes “as code” in a repository (such as Git) that has an immutable audit trail. Any undesirable change is traceable back to the human that introduced it, so strict role separation is unnecessary. Because configuration changes are applied in an automated way across each environment, identical changes can be evaluated in lower environments (such as dev or test) before they are promoted to production, to ensure that no unintended consequences manifest themselves. If problems are discovered, reversion is as easy as identifying the problematic commit and temporarily bypassing it.

In a perfect world, neither developers nor ops staff would have administrative privileges in the production environment. Instead, all changes would be made through an automated, tracked process that has appropriate privileges of its own. Although this is a worthy aspirational goal, our experience has been that it is not yet realistic for most organizations. Work toward this utopian fantasy, but don’t get trapped by it.
31.5 DISASTER MANAGEMENT

Your organization depends on a working IT environment. Not only are you responsible for day-to-day operations, but you should also have plans in place to deal with any reasonably foreseeable eventuality. Preparation for such large-scale problems influences both your overall game plan and the way that you define daily operations. In this section, we look at various kinds of disasters, the data you need to recover gracefully, and the important elements of recovery plans.
Risk assessment

Before designing a disaster recovery plan, it’s a good idea to pull together a risk assessment to help you understand what assets you have, what risks they face, and what mitigation steps you already have in place. The NIST 800-30 special publication details an extensive risk assessment process. You can download it from nist.gov.

Part of the risk assessment process is to make an explicit, written catalog of the potential disasters you want to protect against. Disasters are not all the same, and you may need several different plans to cover the full range of possibilities. For example, some common threat categories are

- Malicious users, both external and internal
- Floods
- Fires
- Earthquakes
- Hurricanes and tornadoes
- Electrical storms and power spikes
- Power failures, both short- and long-term
- Extreme heat or failure of cooling equipment
- ISP/Telecom/Cloud outage
- Device hardware failures (dead servers, fried hard disks)
- Terrorism
- Zombie apocalypse
- Network device failures (routers, switches, cables)
- Accidental user errors (deleted or damaged files and databases, lost configuration information, lost passwords, etc.)

Historically, about half of security breaches originate with insiders. Internal misbehavior continues to be the disaster of highest likelihood at most sites. For each potential threat, consider and write down all the possible implications of that event.

Once you understand the threats, prioritize the services within your IT environment. Build a table that lists your IT services and assigns a priority to each. For example, a “software as a service”
company might rate its external web site as a top-priority service, while an office with a simple, informational external web site might not worry about the site’s fate during a disaster.
Recovery planning

More and more, organizations are designing their critical systems to automatically fail over to secondary servers in the case of problems. This is a great idea if you have little or no tolerance for services being down. However, don’t fall prey to the belief that because you are mirroring your data, you do not need off-line backups. Even if your data centers are miles apart, it is certainly possible that you could lose both of them.

Malicious hackers and ransomware can easily destroy an organization that does not maintain read-only, off-line backups. Make sure you include data backups in your disaster planning.

Read more about cloud computing in Chapter 9.

Cloud computing is often an essential element of disaster planning. Through services such as Amazon’s EC2, you can get a remote site set up and functioning within minutes without having to pay for dedicated hardware. You pay only for what you use, when you use it.

A disaster recovery plan should include the following sections (derived from the NIST disaster recovery standard, 800-34):

- **Introduction** – purpose and scope of the document
- **Concept of operations** – system description, recovery objectives, information classification, line of succession, responsibilities
- **Notification and activation** – notification procedures, damage assessment procedures, plan activation
- **Recovery** – the sequence of events and procedures required to recover lost systems
- **Return to normal operation** – concurrent processing, reconstituted system testing, return to normal operation, plan deactivation

We are accustomed to communicating and accessing documents through the network. However, these facilities may be unavailable or compromised after an incident. Store all relevant contacts and procedures off-line. Know where to get recent backups and how to make use of them without reference to on-line data.

In all disaster scenarios, you will need access to both on-line and off-line copies of essential information. The on-line copies should, if possible, be kept in a self-sufficient environment, one that has a rich complement of tools, has key sysadmins’ environments, runs its own name server, has a complete local `/etc/hosts` file, has no file-sharing dependencies, and so on.

Here’s a list of handy data to keep in the disaster support environment:
• An outline of the recovery procedure: who to call, what to say
• Service contract phone numbers and customer numbers
• Key local phone numbers: police, fire, staff, boss
• Cloud vendor login information
• Inventory of backup media and the backup schedule that produced them
• Network maps
• Software serial numbers, licensing data, and passwords
• Copies of software installation media (can be kept as ISO files)
• Copies of your systems’ service manuals
• Vendor contact information
• Administrative passwords
• Data on hardware, software, and cloud environment configurations: OS versions, patch levels, partition tables, and the like
• Startup instructions for systems that need to be brought back on-line in a particular order
Staffing for a disaster

Your disaster recovery plan should document who will be in charge in the event of a catastrophic incident. Set up a chain of command and keep the names and phone numbers of the principals off-line. We keep a little laminated card with important names and phone numbers printed in microscopic type. Handy—and it fits in your wallet.

The best person to put in charge may be a sysadmin from the trenches, not the IT director (who is usually a poor choice for this role).

The person in charge must be someone who has the authority and decisiveness to make tough decisions in the context of minimal information (e.g., a decision to disconnect an entire department from the network). The ability to make such decisions, communicate them in a sensible way, and lead the staff through the crisis are probably more important than having theoretical insight into system and network management.

An important but sometimes unspoken assumption made in most disaster plans is that sysadmin staff will be available to deal with the situation. Unfortunately, people get sick, go on vacation, leave for other jobs, and in stressful times may even turn hostile. Consider what you’d do if you needed extra emergency help. (Not having enough sysadmins around can sometimes constitute an emergency in its own right if your systems are fragile or your users unsophisticated.)

You might try forming a sort of NATO pact with a local consulting company that has sharable system administration talent. Of course, you must be willing to share back when your buddies have a problem. Most importantly, don’t operate close to the wire in your daily routine. Hire enough system administrators and don’t expect them to work 12-hour days.
Security incidents

System security is covered in detail in Chapter 27. However, it’s worth mentioning here as well because security considerations impact the vast majority of administrative tasks. No aspect of your site’s management strategy can be designed without due regard for security.

For the most part, Chapter 27 concentrates on ways of preventing security incidents from occurring. However, thinking about how you might recover from a security-related incident is an equally important part of security planning.

Having your web site hijacked is a particularly embarrassing type of break-in. For the sysadmin at a web-hosting company, a hijacking can be a calamitous event, especially when it involves sites that handle credit card data. Phone calls stream in from customers, from the media, from the company VIPs who just saw the news of the hijacking on CNN. Who will take the calls? What should that person say? Who is in charge? What role does each person play? If you are in a high-visibility business, it’s definitely worth thinking through this type of scenario, coming up with some preplanned answers, and perhaps even having a practice session to work out the details.

Sites that accept credit card data have legal requirements to deal with after a hijacking. Make sure your organization’s legal department is involved in security incident planning, and make sure you have relevant contact names and phone numbers to call in a time of crisis.

When CNN or Reddit announces that your web site is down, the same effect that makes highway traffic slow down to look at an accident on the side of the road causes your Internet traffic to increase enormously, often to the point of breaking whatever it was that you just fixed. If your web site cannot handle an increase in traffic of 25% or more, consider having your load-balancing device route excess connections to a server that presents a page that simply says “Sorry, we are too busy to handle your request right now.” Of course, forward-thinking capacity planning that includes auto-scaling into the cloud (see Chapter 9) might avoid this situation altogether.

Develop a complete incident handling guide to take the guesswork out of managing security problems. See this page for more details on security incident management.
31.6 IT POLICIES AND PROCEDURES

Comprehensive IT policies and procedures serve as the groundwork for a modern IT organization. Policies set standards for users and administrators and foster consistency for everyone involved. More and more, policies require acknowledgement in the form of a signature or other proof that the user has agreed to abide by their contents. Although this may seem excessive to some, it is actually a great way to protect administrators in the long run.

The ISO/IEC 27001:2013 standard is a good basis for constructing your policy set. It interleaves general IT policies with other important elements such as IT security and the role of the Human Resources department. In the next few sections, we discuss the ISO/IEC 27001:2013 framework and highlight some of its most important and useful elements.
The difference between policies and procedures

Policies and procedures are two distinct things, but they are often confused, and the words are sometimes even used interchangeably. This sloppiness creates confusion, however. To be safe, think of them this way:

- Policies are documents that define requirements or rules. The requirements are usually specified at a relatively high level. An example of a policy might be that incremental backups must be performed daily, with total backups being completed each week.

- Procedures are documents that describe how a requirement or rule will be met. So, the procedure associated with the policy above might say something like “Incremental backups are performed with Backup Exec software, which is installed on the server backups01…”

This distinction is important because your policies should not change often. You should review them annually and maybe change one or two pieces. Procedures, on the other hand, evolve continuously as you change your architecture, systems, and configurations.

Some policy decisions are dictated by the software you are running or by the policies of external groups, such as ISPs. Some policies are mandatory if the privacy of your users’ data is to be protected. We call these topics “nonnegotiable policy.”

In particular, we believe that IP addresses, hostnames, UIDs, GIDs, and usernames should all be managed site-wide. Some sites (multinational corporations, for example) are clearly too large to implement this policy, but if you can swing it, site-wide management makes things a lot simpler. We know of a company that enforces site-wide management for 35,000 users and 100,000 machines, so the threshold at which an organization becomes too big for site-wide management must be pretty high.

Other important issues have a larger scope than just your local IT group:

- Handling of security break-ins
- Filesystem export controls
- Password selection criteria
- Removal of logins for cause
- Copyrighted material (e.g., MP3s and movies)
- Software piracy
**Policy best practices**

Several policy frameworks are available, and they cover roughly the same territories. The following topics are examples of those that are typically included in an IT policy set:

- Information security policy
- External party connectivity agreements
- Asset management policy
- Data classification system
- Human Resources security policy
- Physical security policy
- Access control policies
- Security standards for development, maintenance, and new systems
- Incident management policy
- Business continuity management (disaster recovery)
- Data retention standards
- Protection of user privacy
- Regulatory compliance policy
Procedures

Procedures in the form of checklists or recipes can codify existing practice. They are useful both for new sysadmins and for old hands. Better yet are procedures that include executable scripts or are captured in a configuration management tool such as Ansible, Salt, Puppet, or Chef. Over the long term, most procedures should be automated.

Several benefits accrue from standard procedures:

- Tasks are always done in the same way.
- Checklists reduce the likelihood of errors or forgotten steps.
- It’s faster for the sysadmin to work from a recipe.
- Changes are self-documenting.
- Written procedures provide a measurable standard of correctness.

Here are some common tasks for which you might want to set up procedures:

- Adding a host
- Adding a user
- Localizing a machine
- Setting up backups/snapshots for a new machine
- Securing a new machine
- Removing an old machine
- Restarting a complicated piece of software
- Reviving a web site that is not responding
- Upgrading the operating system
- Patching software
- Installing a software package
- Upgrading critical software
- Backing up and restoring files
- Expiring old backups
- Performing emergency shutdowns
Many issues sit squarely between policy and procedure. For example:

- Who can have an account on your network?
- What happens when they leave?

The resolutions of such issues need to be written down so that you can stay consistent and avoid falling prey to the well-known four-year-old’s ploy of “Mommy said no, let’s go ask Daddy!”
31.7 **Service level agreements**

For the IT organization to keep users happy and meet the needs of the enterprise, the exact details of the service being provided must be negotiated, agreed to, and documented in “service level agreements” or SLAs. A good SLA is a tool that sets appropriate expectations and serves as a reference when questions arise. (But remember, IT provides solutions, not roadblocks!)

When something is broken, users want to know when it’s going to be fixed. That’s it. They don’t really care which hard disk or generator broke, or why; leave that information for your managerial reports.

From a user’s perspective, no news is good news. The system either works or it doesn’t, and if the latter, it doesn’t matter why. Our customers are happiest when they don’t even notice that we exist! Sad, but true.

An SLA helps align end users and support staff. A well-written SLA addresses each of the issues discussed in the following sections.
Scope and descriptions of services

This section is the foundation of the SLA because it describes what the organization can expect from IT. Write it in terms that can be understood by nontechnical staff. Some example services might be

- Email
- Chat
- Internet and web access
- File servers
- Business applications
- Authentication

The standards that IT will adhere to when providing these services must also be defined. For example, an availability section would define the hours of operation, the agreed-on maintenance windows, and the expectations regarding the times at which IT staff will be available to provide live support. One organization might decide that regular support should be available from 8:00 a.m. to 6:00 p.m. on weekdays but that emergency support must be available 24/7. Another organization might decide that it needs standard live support available at all times.

Here is a list of issues to consider when documenting your standards:

- Response time
- Service (and response times) during weekends and off-hours
- House calls (support for environments at home)
- Weird (unique or proprietary) hardware
- Upgrade policy (ancient hardware, software, etc.)
- Supported operating systems
- Supported cloud platforms
- Standard configurations
- Data retention
- Special-purpose software

When considering service standards, keep in mind that many users will want to customize their environments (or even their systems) if the software is not nailed down to prevent this. The stereotypical IT response is to forbid all user modifications, but although this policy makes things easier for IT, it isn’t necessarily the best policy for the organization.
Address this issue in your SLAs and try to standardize on a few specific configurations. Otherwise, your goals of easy maintenance and scaling to grow with the organization will meet some serious impediments. Encourage your creative, OS-hacking employees to suggest modifications that they need for their work, and be diligent and generous in incorporating these suggestions into your standard configurations. If you don’t, your users will work hard to subvert your rules.
Queue prioritization policies

In addition to knowing what services are provided, users must also know about the priority scheme used to manage the work queue. Priority schemes always have wiggle room, but try to design one that covers most situations with few or no exceptions. Some priority-related variables are listed below:

- The importance of the service to the overall organization
- The security impact of the situation (has there been a breach?)
- The service level the customer has paid or contracted for
- The number of users affected
- The importance of any relevant deadline
- The loudness of the affected users (squeaky wheels)
- The importance of the affected users (this is a tricky one, but let’s be honest: some people in your organization have more pull than others)

Although all these factors will influence your rankings, we recommend a simple set of rules together with some common sense to deal with the exceptions. We use the following basic priorities:

- Many people cannot work
- One person cannot work
- Requests for improvements

If two or more requests have top priority and the requests cannot be worked on in parallel, we decide which problem to tackle first by assessing the severity of the issues (e.g., email not working makes almost everybody unhappy, whereas the temporary unavailability of a web service might hinder only a few people). Queues at lower priorities are usually handled in a FIFO manner.
Conformance measurements

An SLA needs to define how the organization will measure your success at fulfilling the terms of the agreement. Targets and goals allow the staff to work toward a common outcome and can lay the groundwork for cooperation throughout the organization. Of course, you must make sure you have tools in place to measure the agreed-on metrics.

At a minimum, you should track the following metrics for your IT infrastructure:

- Percentage or number of projects completed on time and on budget
- Percentage or number of SLA elements fulfilled
- Uptime percentage by system (e.g., “email 99.92% available through Q1”)
- Percentage or number of tickets that were satisfactorily resolved
- Average time to ticket resolution
- Time to provision a new system
- Percentage or number of security incidents handled according to the documented incident handling process
31.8 Compliance: Regulations and Standards

IT auditing and governance are big issues today. Regulations and quasi-standards for specifying, measuring, and certifying compliance have spawned myriad acronyms: SOX, ITIL, COBIT, and ISO 27001, just to name a few. Unfortunately, this alphabet soup is leaving something of a bad taste in system administrators’ mouths, and quality software to implement all the controls deemed necessary by recent legislation is currently lacking.

Some of the major advisory standards, guidelines, industry frameworks, and legal requirements that might apply to system administrators are listed below. The legislative requirements are largely specific to the United States.

Typically, the standard you must use is mandated by your organization type or the data you handle. In jurisdictions outside the United States, you will need to identify the applicable regulations.

- The **CJIS (Criminal Justice Information Systems)** standard applies to organizations that track criminal information and integrate that information with the FBI’s databases. Its requirements can be found on-line at [fbi.gov/hq/cjisd/cjis.htm](http://fbi.gov/hq/cjisd/cjis.htm).

- **COBIT** is a voluntary framework for information management that attempts to codify industry best practices. It is developed jointly by the Information Systems Audit and Control Association (ISACA) and the IT Governance Institute (ITGI); see [isaca.org](http://isaca.org) for details. COBIT’s mission is “to research, develop, publicize, and promote an authoritative, up-to-date, international set of generally accepted information technology control objectives for day-to-day use by business managers and auditors.”

  The first edition of the framework was published in 1996, and we are now at version 5.0, published in 2012. This latest iteration was strongly influenced by the requirements of the Sarbanes-Oxley Act. It includes 37 high-level processes categorized into five domains: Align, Plan, and Organize (APO); Build, Acquire, and Implement (BAI); Deliver, Service, and Support (DSS); Monitor, Evaluate, and Assess (MEA); and Evaluate, Direct, and Monitor (EDM).

- **COPPA**, the **Children’s Online Privacy Protection Act**, regulates organizations that collect or store information about children under the age of 13. Parental permission is required to gather certain information; see [ftc.gov](http://ftc.gov) for details.

- **FERPA**, the **Family Educational Rights and Privacy Act**, applies to all institutions that are recipients of federal aid administered by the Secretary of Education. This regulation protects student information and accords students specific rights with respect to their data. For details, search for FERPA at [ed.gov](http://ed.gov).
- **FISMA**, the **Federal Information Security Management Act**, applies to all government agencies and their contractors. It’s a large and rather vague set of requirements that seek to enforce compliance with a variety of IT security publications from NIST, the National Institute of Standards and Technology. Whether or not your organization falls under the mandate of FISMA, the NIST documents are worth reviewing. See nist.gov for more information.

- The FTC’s **Safe Harbor** framework bridges the gap between the U.S. and E.U. approaches to privacy legislation and defines a way for U.S. organizations that interface with European companies to demonstrate their data security. See export.gov/safeharbor.

- **GLBA**, the **Gramm-Leach-Bliley Act** regulates financial institutions’ use of consumers’ private information. If you’ve been wondering why the world’s banks, credit card issuers, brokerages, and insurers have been pelting you with privacy notices, that’s the Gramm-Leach-Bliley Act at work. See ftc.gov for details. Currently, the best GLBA information is in the business section of the Tips & Advice portion of the web site. The shortcut goo.gl/vv2011 currently works as a deep link.

- **HIPAA**, the **Health Insurance Portability and Accountability Act**, applies to organizations that transmit or store protected health information (aka PHI). It is a broad standard that was originally intended to combat waste, fraud, and abuse in health care delivery and health insurance, but it is now used to measure and improve the security of health information as well. See hhs.gov/ocr/privacy/index.html.

- **ISO 27001:2013** and **ISO 27002:2013** are a voluntary (and informative) collection of security-related best practices for IT organizations. See iso.org.

- **CIP (Critical Infrastructure Protection)** is a family of standards from the North American Electric Reliability Corporation (NERC) which promote the hardening of infrastructure systems such as power, telephone, and financial grids against risks from natural disasters and terrorism. In a textbook demonstration of the Nietzschean concept of organizational “will to power,” it turns out that most of the economy falls into one of NERC’s 17 “critical infrastructure and key resource” (CI/KR) sectors and is therefore richly in need of CIP guidance. Organizations within these sectors should be evaluating their systems and protecting them as appropriate. See nerc.com.

- The **Payment Card Industry Data Security Standard (PCI DSS)** was created by a consortium of payment brands including American Express, Discover, MasterCard, JCB, and Visa. It covers the management of payment card data and is relevant for any organization that accepts credit card payments. The standard comes in two flavors: a self-assessment for smaller organizations and a third-party audit for organizations that process more transactions. See pcisecuritystandards.org.
• The FTC’s **Red Flag Rules** require anyone who extends credit to consumers (i.e., any organization that sends out bills) to implement a formal program to prevent and detect identity theft. The rules require credit issuers to develop heuristics for identifying suspicious account activity; hence, “red flag.” Search for “red flag” at ftc.gov for details.

• In the 1990s and early 2000s, the **Information Technology Infrastructure Library (ITIL)** was a de facto standard for organizations seeking a comprehensive IT service management solution. Many large organizations deployed a formal ITIL program complete with project managers for each process, managers for the project managers, and reporting for managers of the project managers. In most cases, the results were not favorable. The heavy process focus combined with siloed functions resulted in intractable IT constipation. This red tape created opportunities for lean startups to steal market share from well-established companies, thus sending many career IT practitioners out to pasture. We hope we’ve seen the last of ITIL. Some say DevOps is the anti-ITIL methodology.

• Last, but certainly not least, the **IT general controls (ITGC)** portion of the **Sarbanes-Oxley Act (SOX)** applies to all public companies and is designed to protect shareholders from accounting errors and fraudulent practices. See sec.gov.

Some of these standards contain good advice even for organizations that are not required to adhere to them. It might be worth breezing through a few of them just to see if they contain any best practices you might want to adopt. If you have no other constraints, check out NERC CIP and NIST 800-53; they are our favorites with regard to thoroughness and applicability to a broad range of situations.

The National Institute for Standards and Technology (NIST) publishes a host of standards that are useful to administrators and technologists. The two most commonly used ones are mentioned below, but if you are ever bored and looking for standards, you might check out their web site. You will not be disappointed.

NIST 800-53, **Recommended Security Controls for Federal Information Systems and Organizations**, describes how to assess the security of information systems. If your organization has developed an in-house application that holds sensitive information, NIST 800-53 can help you make sure you have truly secured it. Beware, however: embarking on a NIST 800-53 compliance journey is not for the faint of heart. You are likely to end up with a document that is close to 100 pages long and that includes excruciating details. If you plan to do business with a U.S. government agency, you may be required to complete a NIST 800-53 assessment whether you want to or not.

NIST 800-34, **Contingency Planning Guide for Information Technology Systems**, is NIST’s disaster recovery bible. It is directed at government agencies, but any organization can benefit from it. Following the NIST 800-34 planning process takes time, but it forces you to answer
important questions such as, “Which systems are the most critical?”, “How long can we survive without these systems?”, and “How are we going to recover if our primary data center is lost?”
31.9 Legal Issues

The U.S. federal government and several states have enacted laws regarding computer crime. At the federal level, two pieces of legislation date from the early 1990s and three are more recent:

- The Electronic Communications Privacy Act
- The Computer Fraud and Abuse Act
- The No Electronic Theft Act
- The Digital Millennium Copyright Act
- The Email Privacy Act
- The Cybersecurity Act of 2015

Some major issues in the legal arena are these: liability of sysadmins, network providers, and public clouds; peer-to-peer file-sharing networks; copyright issues; and privacy issues. The topics in this section comment on these issues and a variety of other legal debacles related to system administration.
Privacy

Privacy has always been difficult to safeguard, and with the rise of the Internet, it is in more danger than ever. Medical records have been repeatedly disclosed from poorly protected systems, stolen laptops, and misplaced backup tapes. Databases full of credit card numbers are routinely compromised and sold on the black market. Web sites purporting to offer antivirus software actually install spyware when used. Fake email arrives almost daily, appearing to be from your bank and alleging that problems with your account require you to “verify” your account data.

Technical measures can never protect against these attacks because they target your site’s most vulnerable weakness: its users. Your best defense is a well-educated user base. To a first approximation, no legitimate email or web site will ever

- Suggest that you have won a prize;
- Request that you “verify” account information or passwords;
- Ask you to forward a piece of email;
- Ask you to install software you have not explicitly searched for; or
- Inform you of a virus or other security problem.

Users who have a basic understanding of these dangers are more likely to make sensible choices when a pop-up window claims they have won a free MacBook.
Policy enforcement

Log files might prove to you conclusively that person X did bad thing Y, but to a court it is all just hearsay evidence. Protect yourself with written policies. Log files sometimes include time stamps, which are useful but not necessarily admissible as evidence unless you can also prove that the computer was running the Network Time Protocol (NTP) to keep its clock synced to a reference standard.

You may need a security policy to prosecute someone for misuse. That policy should include a statement such as this: “Unauthorized use of computing systems may involve not only transgression of organizational policy but also a violation of state and federal laws. Unauthorized use is a crime and may involve criminal and civil penalties; it will be prosecuted to the full extent of the law.”

We advise you to display a splash screen that advises users of your snooping policy. You might say something like: “Activity may be monitored in the event of a real or suspected security incident.”

To ensure that users see the notification at least once, include it in the startup files you give to new users. If you require the use of SSH to log in (and you should), you can configure /etc/ssh/sshd_config so that SSH always shows the splash screen.

Be sure to specify that through the act of using their accounts, users acknowledge your written policy. Explain where users can get additional copies of policy documents, and post key documents on an appropriate web page. Also include the specific penalty for noncompliance (e.g., deletion of the account).

In addition to displaying the splash screen, have users sign a policy agreement before giving them access to your systems. Craft the acceptable use agreement in conjunction with your legal department. If you don’t have signed agreements from current employees, make a sweep to obtain them, then make signing the agreement a standard part of the induction process for new hires.

You might also consider periodically offering training sessions on information security. This is a great opportunity to educate users about important issues such as phishing scams, when it’s OK to install software and when it’s not, password security, and any other points that affect your environment.
Control = liability

Service providers (ISP, cloud, etc.) typically have an appropriate use policy (AUP) dictated by their upstream providers and required of their downstream customers. This “flow down” of liability assigns responsibility for users’ actions to the users themselves, not to the service provider or the service provider’s upstream provider. Such policies have been used to attempt spam control and to protect service providers in cases where customers have stored illegal or copyrighted material in their accounts. Check the laws in your area; your mileage may vary.

Your policies should explicitly state that users are not to use organizational resources for illegal activities. However, that’s not really enough—you also need to discipline users if you find out they are misbehaving. Organizations that know about violations but do not act on them are complicit and can be prosecuted. Unenforced or inconsistent policies are worse than none, from both a practical and legal point of view.

Because of the risk of being found complicit in user activities, some sites limit the data that they log, the length of time for which log files are kept, and the amount of log file history kept on backup tapes. Some software packages help with the implementation of this policy by including levels of logging that help the sysadmin debug problems but that do not violate users’ privacy. However, always be aware of what kind of logging might be required by local laws or by any regulatory standards that apply to you.
Software licenses

Many sites have paid for K copies of a software package and have N copies in daily use, where K < N. Getting caught in this situation could be damaging to the company—probably more damaging than the cost of those N-minus-K other licenses. Other sites have received a demo copy of an expensive software package and hacked it (reset the date on the machine, found a license key, etc.) to make it continue working after the expiration of the demo period. How do you as a sysadmin deal with requests to violate license agreements and make copies of software on unlicensed machines? What do you do when you find that machines for which you are responsible are running pirated software?

It’s a tough call. Management will often not back you up in your requests that unlicensed copies of software be either removed or paid for. Often, it is a sysadmin who signs the agreement to remove the demo copies after a certain date, but a manager who makes the decision not to remove them.

We are aware of several cases in which a sysadmin’s immediate manager would not deal with the situation and told the sysadmin not to rock the boat. The admin then wrote a memo to the boss asking for correction of the situation and documenting the number of copies of the software that were licensed and the number that were in use. The admin quoted a few phrases from the license agreement and carbon copied the president of the company and his boss’s managers. In one case, this procedure worked and the sysadmin’s manager was let go. In another case, the sysadmin quit when even higher management refused to do the right thing. No matter what you do in such a situation, get things in writing. Ask for a written reply; if all you get is spoken words, write a short memo documenting your understanding of your instructions and send it to the person in charge.
31.10 Organizations, Conferences, and Other Resources

Many UNIX and Linux support groups—both general and vendor-specific—help you network with other people who are running the same software. Table 31.3 briefly lists a few such organizations, but many other national and regional groups are not listed here.

Table 31.3: UNIX and Linux organizations of interest to system administrators

<table>
<thead>
<tr>
<th>Name</th>
<th>What it is</th>
</tr>
</thead>
<tbody>
<tr>
<td>FSF</td>
<td>The Free Software Foundation, sponsor of GNU</td>
</tr>
<tr>
<td>USENIX</td>
<td>UNIX/Linux user group, quite technically oriented(^a)</td>
</tr>
<tr>
<td>LOPSA</td>
<td>The League of Professional System Administrators</td>
</tr>
<tr>
<td>SANS</td>
<td>Sponsors sysadmin and security conferences</td>
</tr>
<tr>
<td>SAGE-AU</td>
<td>Australian sysadmins who hold yearly conferences in Oz</td>
</tr>
<tr>
<td>Linux Foundation</td>
<td>Nonprofit Linux consortium; produces LinuxCon among others</td>
</tr>
<tr>
<td>LinuxFest Northwest</td>
<td>Grass-roots conference with great content</td>
</tr>
</tbody>
</table>

\(^a\) Well-known parent organization of the LISA special interest group, which was retired in 2016

FSF, the Free Software Foundation, sponsors the GNU Project ("GNU’s Not Unix,” a recursive acronym). The “free” in the FSF’s name is the “free” of free speech and not that of free beer. The FSF is also the origin of the GNU Public License, which now exists in several versions and covers many of the free software packages used on UNIX and Linux systems.

USENIX, an organization of users of Linux, UNIX, and other open source operating systems, holds one general conference and several specialized (smaller) conferences or workshops each year. The Annual Technical Conference (ATC) is a potpourri of in-depth UNIX and Linux topics and is a great place for networking with the community.

The League of Professional System Administrators, LOPSA, has a fairly complex and somewhat sordid history. It was originally associated with USENIX and was intended to assume the mantle of USENIX’s system administration special interest group, SAGE. Unfortunately, LOPSA and USENIX parted on less than amicable terms and are now separate organizations.

Today, LOPSA sponsors a variety of sysadmin-related networking, mentorship, and educational programs, including events such as System Administrator Appreciation Day on the last Friday of July. The customary gift for this holiday is bottle of scotch.

SANS offers courses and seminars in the security space and also founded a certification program, the Global Information Assurance Certification (GIAC), which operates somewhat independently. Certifications are available in a variety of general and specific skill areas such as system administration, coding, incident handling, and forensics. See giac.org for details.
Many local areas have their own regional UNIX, Linux, or open systems user groups. Meetup.com is an excellent resource for finding relevant groups in your area. Local groups usually have regular meetings, workshops with local or visiting speakers, and often, dinner together before or after the meetings. They’re a good way to network with other sysadmins.
31.11 RECOMMENDED READING


MORRIS, KIEF. Infrastructure as Code: Managing Servers in the Cloud. Sebastopol, CA: O’Reilly Media, 2016. This book is a well-written 10,000-foot overview of DevOps and large-scale tools for system administration in the cloud. It includes few specifics about configuration management per se, but it’s helpful for understanding how configuration management integrates into the larger scheme of DevOps and structured administration.

The site itl.nist.gov is the landing page for the NIST Information Technology Laboratory and includes lots of information about standards.

The web site of the Electronic Frontier Foundation, eff.org, is a great place to find commentary on the latest issues in privacy, cryptography, and legislation. Always interesting reading.

SANS hosts a collection of security policy templates at sans.org/resources/policies.
In the modern age, most folks have at least a vague idea what system administrators do: work tirelessly to meet the needs of their users and organizations, plan and implement a robust computing environment, and pull proverbial rabbits out of many different hats. Although sysadmins are often viewed as underpaid and underappreciated, most users can at least identify their friendly local sysadmin—in many cases, more quickly than they can name their boss’s boss.

It wasn’t always this way. Over the last 50 years (and the 30-year history of this book), the role of the system administrator has evolved hand-in-hand with UNIX and Linux. A full understanding of system administration requires an understanding of how we got here and of some of the historical influences that have shaped our landscape. Join us as we reflect on the many wonderful years.

**THE DAWN OF COMPUTING: SYSTEM OPERATORS (1952–1960)**

The first commercial computer, the IBM 701, was completed in 1952. Before the 701, all computers had been one-offs. In 1954, a redesigned version of the 701 was announced as the IBM 704. It had 4,096 words of magnetic core memory and three index registers. It used 36-bit words (as opposed to the 701’s 18-bit words) and did floating-point arithmetic. It executed 40,000 instructions every second.
But the 704 was more than just an update: it was incompatible with the 701. Although deliveries were not to begin until late 1955, the operators of the eighteen 701s in existence (the predecessors of modern system administrators) were already fretful. How would they survive this “upgrade,” and what pitfalls lay ahead?

IBM itself had no solution to the upgrade and compatibility problem. It had hosted a “training class” for customers of the 701 in August 1952, but there were no textbooks. Several people who had attended the training class continued to meet informally and discuss their experiences with the system. IBM encouraged the operators to meet, to discuss their problems, and to share their solutions. IBM funded the meetings and made available to the members a library of 300 computer programs. This group, known as SHARE, is still the place (60+ years later) where IBM customers meet to exchange information. (Although SHARE was originally a vendor-sponsored organization, today it is independent.)

FROM SINGLE-PURPOSE TO TIME SHARING (1961–1969)

Early computing hardware was physically large and extraordinarily expensive. These facts encouraged buyers to think of their computer systems as tools dedicated to some single, specific mission: whatever mission was large enough and concrete enough to justify the expense and inconvenience of the computer.

If a computer were a single-purpose tool—let’s say, a saw—then the staff that maintained that computer would be the operators of the saw. Early system operators were viewed more as “folks that cut lumber” than as “folks that provide what’s necessary to build a house.” The transition from system operator to system administrator did not start until computers began to be seen as multipurpose tools. The advent of time sharing was a major reason for this change in viewpoint.

John McCarthy had begun thinking about time sharing in the mid-1950s. But it was only at MIT (in 1961–62) that he, Jack Dennis, and Fernando Corbato talked seriously about permitting “each user of a computer to behave as though he were in sole control of a computer.”

In 1964, MIT, General Electric, and Bell Labs embarked on a project to build an ambitious time-sharing system called Multics, the Multiplexed Information and Computing Service. Five years later, Multics was over budget and far behind schedule. Bell Labs pulled out of the project.

UNIX IS BORN (1969–1973)

Bell Labs’ abandonment of the Multics project left several researchers in Murray Hill, NJ, with nothing to work on. Three of them—Ken Thompson, Rudd Canaday, and Dennis Ritchie—had liked certain aspects of Multics but hadn’t been happy with the size and the complexity of the system, and they often gathered in front of a whiteboard to delve into design philosophy. The Labs had Multics running on its GE-645, and Thompson continued to work on it “just for fun.”
Doug McIlroy, the manager of the group, said, “When Multics began to work, the very first place it worked was here. Three people could overload it.”

In the summer of 1969, Thompson became a temporary bachelor for a month when his wife, Bonnie, took their year-old son to meet his relatives on the West Coast. Thompson recalled, “I allocated a week each to the operating system, the shell, the editor, and the assembler…it was totally rewritten in a form that looked like an operating system, with tools that were sort of known; you know, assembler, editor, shell—if not maintaining itself, right on the verge of maintaining itself, to totally sever the GECOS connection…essentially one person for a month.” (GECOS was the General Electric Comprehensive Operating System.)

Steve Bourne, who joined Bell Labs the next year, described the cast-off PDP-7 used by Ritchie and Thompson: “The PDP-7 provided only an assembler and a loader. One user at a time could use the computer…The environment was crude, and parts of a single-user UNIX system were soon forthcoming…[The] assembler and rudimentary operating system kernel were written and cross-assembled for the PDP-7 on GECOS. The term UNICS was apparently coined by Peter Neumann, an inveterate punster, in 1970.” The original UNIX was a single-user system, obviously an “emasculated Multics.” But although there were aspects of UNICS/UNIX that were influenced by Multics, there were also, as Dennis Ritchie said, “profound differences.”

“We were a bit oppressed by the big system mentality,” he said. “Ken wanted to do something simple. Presumably, as important as anything was the fact that our means were much smaller. We could get only small machines with none of the fancy Multics hardware. So, UNIX wasn’t quite a reaction against Multics…Multics wasn’t there for us anymore, but we liked the feel of interactive computing that it offered. Ken had some ideas about how to do a system that he had to work out…Multics colored the UNIX approach, but it didn’t dominate it.”

Ken and Dennis’s “toy” system didn’t stay simple for long. By 1971, user commands included as (the assembler), cal (a simple calendar tool), cat (catenate and print), chdir (change working directory), chmod (change mode), chown (change owner), cmp (compare two files), cp (copy file), date, dc (desk calculator), du (summarize disk usage), ed (editor), and over two dozen others. Most of these commands are still in use.

By February 1973, there were 16 UNIX installations. Two big innovations had occurred. The first was a “new” programming language, C, based on B, which was itself a “cut-down” version of Martin Richards’ BCPL (Basic Combined Programming Language). The other innovation was the idea of a pipe.

A pipe is a simple concept: a standardized way of connecting the output of one program to the input of another. The Dartmouth Time-Sharing System had communication files, which anticipated pipes, but their use was far more specific. The notion of pipes as a general facility was Doug McIlroy’s. The implementation was Ken Thompson’s, at McIlroy’s insistence. (“It was one of the only places where I very nearly exerted managerial control over UNIX,” Doug said.)
“It’s easy to say ‘cat into grep into…’ or ‘who into cat into grep’ and so on,” McIlroy remarked. “It’s easy to say and it was clear from the start that it would be something you’d like to say. But there are all these side parameters… And from time to time I’d say ‘How about making something like this?’ And one day I came up with a syntax for the shell that went along with piping, and Ken said ‘I’m going to do it!’”

In an orgy of rewriting, Thompson updated all the UNIX programs in one night. The next morning there were one-liners. This was the real beginning of the power of UNIX—not from the individual programs, but from the relationships among them. UNIX now had a language of its own as well as a philosophy:

- Write programs that do one thing and do it well.
- Write programs to work together.
- Write programs that handle text streams as a universal interface.

A general-purpose time-sharing OS had been born, but it was trapped inside Bell Labs. UNIX offered the promise of easily and seamlessly sharing computing resources among projects, groups, and organizations. But before this multipurpose tool could be used by the world, it had to escape and multiply. Katy bar the door!


In October 1973, the ACM held its Symposium on Operating Systems Principles (SOSP) in the auditorium at IBM’s new T.J. Watson Research Center in Yorktown Heights, NY. Ken and Dennis submitted a paper, and on a beautiful autumn day, drove up the Hudson Valley to deliver it. (Thompson made the actual presentation.) About 200 people were in the audience, and the talk was a smash hit.

Over the next six months, the number of UNIX installations tripled. When the paper was published in the July 1974 issue of the *Communications of the ACM*, the response was overwhelming. Research labs and universities saw shared UNIX systems as a potential solution to their growing need for computing resources.

According to the terms of a 1958 antitrust settlement, the activities of AT&T (parent of Bell Labs) were restricted to running the national telephone system and to special projects undertaken on behalf of the federal government. Thus, AT&T could not sell UNIX as a product and Bell Labs had to license its technology to others. In response to requests, Ken Thompson began shipping copies of the UNIX source code. According to legend, each package included a personal note signed “love, ken.”

One person who received a tape from Ken was Professor Robert Fabry of the University of California at Berkeley. By January 1974, the seed of Berkeley UNIX had been planted.
Other computer scientists around the world also took an interest in UNIX. In 1976, John Lions (on the faculty of the University of New South Wales in Australia) published a detailed commentary on a version of the kernel called V6. This effort became the first serious documentation of the UNIX system and helped others to understand and expand on Ken and Dennis’s work.

Students at Berkeley enhanced the version of UNIX they had received from Bell Labs to meet their needs. The first Berkeley tape (1BSD, short for 1st Berkeley Software Distribution) included a Pascal system and the vi editor for the PDP-11. The student behind the release was a grad student named Bill Joy. 2BSD came the next year, and 3BSD, the first Berkeley release for the DEC VAX, was distributed in late 1979.

In 1980, Professor Fabry struck a deal with the Defense Advanced Research Project Agency (DARPA) to continue the development of UNIX. This arrangement led to the formation of the Computer Systems Research Group (CSRG) at Berkeley. Late the next year, 4BSD was released. It became quite popular, largely because it was the only version of UNIX that ran on the DEC VAX 11/750, the commodity computing platform of the time. Another big advancement of 4BSD was the introduction of TCP/IP sockets, the generalized networking abstraction that spawned the Internet and is now used by most modern operating systems. By the mid-1980s, most major universities and research institutions were running at least one UNIX system.

In 1982, Bill Joy took the 4.2BSD tape with him to start Sun Microsystems (now part of Oracle America) and the Sun operating system (SunOS). In 1983, the court-ordered divestiture of AT&T began. One unanticipated side effect of the divestiture was that AT&T was now free to begin selling UNIX as a product. They released AT&T UNIX System V, a well-recognized albeit awkward commercial implementation of UNIX.

Now that Berkeley, AT&T, Sun, and other UNIX distributions were available to a wide variety of organizations, the foundation was laid for a general computing infrastructure built on UNIX technology. The same system that was used by the astronomy department to calculate star distances could be used by the applied math department to calculate Mandelbrot sets. And that same system was simultaneously providing email to the entire university.

**The Rise of System Administrators**

The management of general-purpose computing systems demanded a different set of skills than those required just two decades earlier. Gone were the days of the system operator who focused on getting a single computer system to perform a specialized task. System administrators came into their own in the early 1980s as people who ran UNIX systems to meet the needs of a broad array of applications and users.

Because UNIX was popular at universities and because those environments included lots of students who were eager to learn the latest technology, universities were early leaders in the development of organized system administration groups. Universities such as Purdue, the
University of Utah, the University of Colorado, the University of Maryland, and the State University of New York (SUNY) Buffalo became hotbeds of system administration.

System administrators also developed an array of their own processes, standards, best practices, and tools (such as `sudo`). Most of these products were built out of necessity; without them, unstable systems and unhappy users were the result.

Evi Nemeth became known as the “mother of system administration” by recruiting undergraduates to work as system administrators to support the Engineering College at the University of Colorado. Her close ties with folks at Berkeley, the University of Utah, and SUNY Buffalo created a system administration community that shared tips and tools. Her crew, often called the “munchkins” or “Evi slaves” attended USENIX and other conferences and worked as on-site staff in exchange for the opportunity to absorb information at the conference.

It was clear early on that system administrators had to be rabid jacks of all trades. A system administrator might start a typical day in the 1980s by using a wire-wrap tool to fix an interrupt jumper on a VAX backplane. Mid-morning tasks might include sucking spilled toner out of a malfunctioning first-generation laser printer. Lunch hour could be spent helping a grad student debug a new kernel driver, and the afternoon might consist of writing backup tapes and hassling users to clean up their home directories to make space in the filesystem. A system administrator was, and is, a fix-everything, take-no-prisoners guardian angel.

The 1980s were also a time of unreliable hardware. Rather than living on a single silicon chip, the CPUs of the 1980s were made up of several hundred chips, all of them prone to failure. It was the system administrator’s job to isolate failed hardware and get it replaced, quickly. Unfortunately, these were also the days before it was common to FedEx parts on a whim, so finding the right part from a local source was often a challenge.

In one case, our beloved VAX 11/780 was down, leaving the entire campus without email. We knew there was a business down the street that packaged VAXes to be shipped to the (then cold-war) Soviet Union “for research purposes.” Desperate, we showed up at their warehouse with a huge wad of cash in our pocket, and after about an hour of negotiation, we escaped with the necessary board. At the time, someone remarked that it felt more comfortable to buy drugs than VAX parts in Boulder.

**SYSTEM ADMINISTRATION DOCUMENTATION AND TRAINING**

As more individuals began to identify themselves as system administrators—and as it became clear that one might make a decent living as a sysadmin—requests for documentation and training became more common. In response, folks like Tim O’Reilly and his team (then called O’Reilly and Associates, now O’Reilly Media) began to publish UNIX documentation that was based on hands-on experience and written in a straightforward way.

See Chapter 31 for more pointers to sysadmin resources.
As a vehicle for in-person interaction, the USENIX Association held its first conference focused on system administration in 1987. This Large Installation System Administration (LISA) conference catered mostly to a west coast crowd. Three years later, the SANS (SysAdmin, Audit, Network, Security) Institute was established to meet the needs of the east coast. Today, both the LISA and SANS conferences serve the entire U.S. region, and both are still going strong.

In 1989, we published the first edition of this book, then titled *UNIX System Administration Handbook*. It was quickly embraced by the community, perhaps because of the lack of alternatives. At the time, UNIX was so unfamiliar to our publisher that their production department replaced all instances of the string “etc” with “and so on,” resulting in filenames such as `/and so on/passwd`. We took advantage of the situation to seize total control of the bits from cover to cover, but the publisher is admittedly much more UNIX savvy today. Our 30-year relationship with this same publisher has yielded a few other good stories, but we’ll omit them out of fear of souring our otherwise amicable relationship.

**UNIX hugged to near death, Linux is born (1991–1995)**

By late 1990, it seemed that UNIX was well on its way to world domination. It was unquestionably the operating system of choice for research and scientific computing, and it had been adopted by mainstream businesses such as Taco Bell and McDonald’s. Berkeley’s CSRG group, then consisting of Kirk McKusick, Mike Karels, Keith Bostic, and many others, had just released 4.3BSD-Reno, a pun on an earlier 4.3 release that added support for the CCI Power 6/32 (code named “Tahoe”) processor.

Commercial releases of UNIX such as SunOS were also thriving, their success driven in part by the advent of the Internet and the first glimmers of e-commerce. PC hardware had become a commodity. It was reasonably reliable, inexpensive, and relatively high-performance. Although versions of UNIX that ran on PCs did exist, all the good options were commercial and closed source. The field was ripe for an open source PC UNIX.

In 1991, a group of developers that had worked together on the BSD releases (Donn Seeley, Mike Karels, Bill Jolitz, and Trent R. Hein), together with a few other BSD advocates, founded Berkeley Software Design, Inc. (BSDI). Under the leadership of Rob Kolstad, BSDI provided binaries and source code for a fully functional commercial version of BSD UNIX on the PC platform. Among other things, this project proved that inexpensive PC hardware could be used for production computing. BSDI fueled explosive growth in the early Internet as it became the operating system of choice for early Internet service providers (ISPs).

In an effort to recapture the genie that had escaped from its bottle in 1973, AT&T in 1992 filed a lawsuit against BSDI and the Regents of the University of California, alleging code copying and theft of trade secrets. It took AT&T’s lawyers over two years to identify the offending code. When all was said and done, the lawsuit was settled and three files (out of more than 18,000) were removed from the BSD code base.
Unfortunately, this two-year period of uncertainty had a devastating effect on the entire UNIX world, BSD and non-BSD versions alike. Many companies jumped ship to Microsoft Windows, fearful that they would end up at the mercy of AT&T as it hugged its child to near-death. By the time the dust cleared, BSDI and the CSRG were both mortally wounded. The BSD era was coming to an end.

Meanwhile, Linus Torvalds, a Helsinki college student, had been playing with Minix (a PC-based UNIX clone developed by Andrew S. Tanenbaum) and began writing his own UNIX clone. By 1992, a variety of Linux distributions (including SuSE and Yggdrasil Linux) had emerged. 1994 saw the establishment of Red Hat and Linux Pro.

Multiple factors have contributed to the phenomenal success of Linux. The strong community support enjoyed by the system and its vast catalog of software from the GNU archive make Linux quite a powerhouse. It works well in production environments, and some folks argue that you can build a more reliable and performant system on top of Linux than you can on top of any other operating system. It’s also interesting to consider that part of Linux’s success may relate to the golden opportunity created for it by AT&T’s action against BSDI and Berkeley. That ill-timed lawsuit struck fear into the hearts of UNIX advocates right at the dawn of e-commerce and the start of the Internet bubble.

But who cares, right? What remained constant through all these crazy changes was the need for system administrators. A UNIX system administrator’s skill set is directly applicable to Linux, and most system administrators guided their users gracefully through the turbulent seas of the 1990s. That’s another important characteristic of a good system administrator: calm during a storm.

A WORLD OF WINDOWS (1996–1999)

Microsoft first released Windows NT in 1993. The release of a “server” version of Windows, which had a popular user interface, generated considerable excitement just as AT&T was busy convincing the world that it might be out to fleece everyone for license fees. As a result, many organizations adopted Windows as their preferred platform for shared computing during the late 1990s. Without question, the Microsoft platform has come a long way, and for some organizations it is the best option.

Unfortunately, UNIX, Linux, and Windows administrators initially approached this marketplace competition in an adversarial stance. “Less filling” vs. “tastes great” arguments erupted in organizations around the world. (Just for the record, Windows is indeed less filling.) Many UNIX and Linux system administrators started learning Windows, convinced they’d be put out to pasture if they didn’t. After all, Windows 2000 was on the horizon. By the close of the millennium, the future of UNIX looked grim.

UNIX AND LINUX THRIVE (2000–2009)
As the Internet bubble burst, everyone scrambled to identify what was real and what had been only a venture-capital-fueled mirage. As the smoke drifted away, it became clear that many organizations with successful technology strategies were using UNIX or Linux along with Windows rather than one or the other. It wasn’t a war anymore.

A number of evaluations showed that the total cost of ownership (TCO) of a Linux server was significantly lower than that of a Windows server. As the impact of the 2008 economic crash hit, TCO became more important than ever. The world again steered toward open source versions of UNIX and Linux.

UNIX AND LINUX IN THE HYPERSCALE CLOUD (2010-PRESENT)

Linux and PC-based UNIX variants such as FreeBSD have continued to expand their market share, with Linux being the only operating system whose market share on servers is growing. Not to be left out, Apple’s current full-size operating system, macOS, is also a variant of UNIX. (Even Apple’s iPhone runs a cousin of UNIX, and Google’s Android operating system derives from the Linux kernel.)

Much of the recent growth in UNIX and Linux has occurred in the context of virtualization and cloud computing. See Chapter 24, Virtualization, and Chapter 9, Cloud Computing, for more details about these technologies.

The ability to create virtual infrastructure (and entire virtual data centers) by making API calls has fundamentally shifted the course of the river once again. Gone are the days of managing physical servers by hand. Scaling infrastructure no longer means slapping down a credit card and waiting for boxes to appear on the loading dock. Thanks to services such as Google GCP, Amazon AWS, and Microsoft Azure, the era of the hyperscale cloud has arrived. Standardization, tools, and automation are not just novelties but intrinsic attributes of every computing environment.

Today, competent management of fleets of servers requires extensive knowledge and skill. System administrators must be disciplined professionals. They must know how to build and scale infrastructure, how to work collaboratively with peers in a DevOps environment, how to code simple automation and monitoring scripts, and how to remain calm when a thousand servers are down at once. (One thing hasn’t changed: whiskey is still a close friend to many system administrators.)

UNIX AND LINUX TOMORROW

Where are we headed next? The lean, modular paradigm that has served UNIX so well over the last few decades is also one foundation of the up-and-coming Internet of Things (IoT). The Brookings Institution estimates that 50 billion small, distributed IoT devices will exist by 2020 (see brook.gs/2bNwbya).
It’s tempting to think of these devices as we thought of the non-networked consumer appliances (e.g., toaster ovens or blenders) of yesteryear: plug them in, use them for a few years, and if they break, throw them in the landfill. They don’t need “management” or central administration, right?

In fact, nothing could be further from the truth. Many of these devices handle sensitive data (e.g., audio streamed from a microphone in your living room) or perform mission-critical functions such as controlling the temperature of your house.

Some of these devices run embedded software derived from the OSS world. But regardless of what’s inside the devices themselves, the majority report back to a mother ship in the cloud that runs—you guessed it—UNIX or Linux. In the early market-share land grab, many devices have already been deployed without much thought to security or to how the ecosystem will operate in the future.

The IoT craze isn’t limited to the consumer market. Modern commercial buildings are riddled with networked devices and sensors for lighting, HVAC, physical security, and video, just to name a few. These devices often pop up on the network without coordination from the IT or Information Security departments. They’re then forgotten without any plan for ongoing management, patching, or monitoring.

Size doesn’t matter when it comes to networked systems. System administrators need to advocate for the security, performance, and availability of IoT devices (and their supporting infrastructure) regardless of size, location, or function.

System administrators hold the world’s computing infrastructure together, solve the hairy problems of efficiency, scalability, and automation, and provide expert technology leadership to users and managers alike.

We are system administrators. Hear us roar!

**Recommended reading**


**Salus, Peter H.** *Casting the Net: From ARPANET to Internet and Beyond.* Reading, MA: Addison-Wesley, 1995.

**Salus, Peter H.** *The Daemon, the Gnu, and the Penguin.* Marysville, WA: Reed Media Services, 2008. This book was also serialized at www.groklaw.net.
Colophon

For previous editions of this book, we used Adobe FrameMaker as our layout tool and were (more or less) happy with its features, stability, and facility with book-length writing projects. But although FrameMaker still exists, it now runs only on Windows and has become an increasingly vestigial member of the Adobe lineup.

Rather than targeting the general publishing market, Adobe seems to be pitching FrameMaker predominantly to multilingual workshops and to the poor lost souls who generate government-mandated SGML content. The product’s core features haven’t been updated in decades. It’s like seeing a much-admired football quarterback working at the local mini mart loading hot dogs onto the fancy rolling cooker: someone has to do the job, but it seems like a waste of talent.

Ever since Adobe InDesign debuted in 2000, we’ve been waiting for it to grow sufficiently mature that we could contemplate switching. It was never originally intended for long documents, but over the years, the addition of features like tables of contents and indexing suggested that Adobe hoped to develop InDesign as a general solution for document publishing.

For this edition, we finally allowed ourselves to be goaded into the InDesign corral. We ran Adobe InDesign CC 2017 on macOS systems. Hilarity ensued…at least, if you find slapstick hilarious.

InDesign has many strengths and adherents, and we’ll certainly reach for it the next time we need to design a six-page glossy brochure. However, we’re reasonably certain that most InDesign fans aren’t writing technical books.

One author said it best: “I have the impression that InDesign’s book features were designed primarily to let InDesign slip past product selection committees’ screening phases.” To get this book into your hands, we’ve had to write literally thousands of lines of InDesign JavaScript code. Some of that code is ULSAH-specific, but a lot of it does basic book production chores that FrameMaker handled out of the box.

Unfortunately, our complaints with InDesign go beyond its cursory support for book production. We’ve also been continually surprised by its instability, obvious bugs, and nondeterministic
behaviors.

We wish we had some useful tooling advice to impart to other book authors, but we’re just as puzzled as everyone else. FrameMaker has no future and InDesign has no present. Second-tier and on-line book publishing systems seem largely to target those looking for EZ-mode, book-length text editors. As far as we can tell, no modern GUI software addresses the task of creating a book like this one.

Ah well, there’s always Microsoft Word. Our publisher tells us it’s still the predominant format in which authors submit manuscripts.

We used the IndexUtilities add-on from Kerntiff Publishing Systems (kerntiff.co.uk) to supplement InDesign’s fledgling indexing system. *Indexing from A to Z* by Hans H. Wellisch remains an invaluable reference text for the art of indexing and a resource that we recommend highly.

Lisa Haney drew the interior cartoons with a 0.05mm Staedtler pigment liner, then scanned them and converted them to 1200dpi bitmaps. The cover artwork was executed on black Ampersand Clayboard (a scratchboard) with Dr. Martin’s Dyes for color. After scanning, the cover art was color-corrected in Photoshop and the layout completed in Adobe Illustrator.

The body text is Minion Pro, designed by Robert Slimbach. Headings, tables, and illustrations are set in Myriad Pro SemiCondensed by Robert Slimbach and Carol Twombly, with Fred Brady and Christopher Slye.

We’ve long sought a good typographical solution for “code” samples, and after trying out pretty much every font and option on the market, we’ve finally found our one true love: the Input font system by David Jonathan Ross, available from input.fontbureau.com. It looks great, and with 168 different variants, it’s easy to match to any given body text.

Better yet, Input has both monospaced and proportionally spaced versions. You can use well-behaved proportional variants for most purposes, then switch to monospaced for tabular output. The styles intermix cleanly, so readers won’t even notice a difference unless they closely scrutinize the typography.

The authors were never in the same physical location during this project. We maintained a shared tree of source files hosted on GitLab. Binary .indd (InDesign) files were stored in the repository and managed with the help of a horde of custom Ruby scripts. This approach was tolerable, but the inability to see per-commit diffs for text changes (because of InDesign’s binary file format) limited its flexibility.

InDesign does support an XML interchange format, but unfortunately, every trip to XML resets all internal object IDs. The same file saved out twice in XML appears to have thousands of differences. Hence, no diffs, no merging, and no collaboration.
James Garnett holds a PhD in Computer Science from the University of Colorado and is a Senior Software Engineer at Secure64 Software, Inc., where he develops DDoS mitigation technologies for Linux kernels. When not knee-deep in kernel code, he is usually somewhere deep in the Cascade Mountain range of Washington state.

Fabrizio Branca (@fbrnc) is the Lead System Developer at AOE. He, his wife, and their two children just returned to Germany after living in San Francisco for four years. Fabrizio has contributed to several open source projects. He focuses on architecture, infrastructure, and high-performance applications. He promotes solid development, testing, and deployment processes for large projects.

Adrian Mouat (@adrianmouat) has been involved with containers from the early days of Docker and wrote the O’Reilly book Using Docker (amzn.to/2sVAIZ). He is currently Chief Scientist at Container Solutions, a pan-European company focusing on consulting and product development for microservices and containers.
About the Authors

For general comments and bug reports, please contact ulsah@book.admin.com. We regret that we are unable to answer technical questions.

Evi Nemeth retired from the Computer Science faculty at the University of Colorado in 2001. She explored the Pacific on her 40-foot sailboat named Wonderland for many years before becoming lost at sea in 2013 (see this page). The fourth edition of this book was her last as an active participant, but we’re proud to say that we’ve retained her writing where possible.

Garth Snyder (@GarthSnyder) has worked at NeXT and Sun and holds a BS in Engineering from Swarthmore College and an MD and an MBA from the University of Rochester.

Trent R. Hein (@trenthein) is a serial entrepreneur who is passionate about practical cybersecurity and automation. Outside of technology, he loves hiking, skiing, fly fishing, camping, bluegrass, dogs, and the Oxford comma. Trent holds a BS in Computer Science from the University of Colorado.

Ben Whaley is the founder of WhaleTech, an independent consultancy. He was honored by Amazon as one of the first AWS Community Heroes. He obtained a BS in Computer Science from the University of Colorado at Boulder.

Dan Mackin (@dan_mackin) has a BS in Electrical and Computer Engineering from the University of Colorado at Boulder. He applies Linux and other open source technologies not only to his work, but also to automation, monitoring, and weather metrics collection projects at home. Dan loves skiing, sailing, backcountry touring, and spending time with his wife and dog.
We have alphabetized files under their last components. And in most cases, only the last component is listed. For example, to find index entries relating to the /etc/mail/aliases file, look under aliases. Our friendly vendors have forced our hand by hiding standard files in new and inventive directories on each system.

Symbols

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802.2* IEEE standards see IEEE standards

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