
Quantum Computing and Quantum Information

An Expository Manuscript

A Symphony of Theoretical Computer Science and Quantum Physics

by
S. Padmapriya and Nishkal Rao

Indian Institute of Science Education and Research (IISER) Pune

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To our parents and teachers

Preface

Teaching is the best form of learning, and when passion meets perseverance, surprising results occur. This work is one such result of a passion project in the pursuit of knowledge and love for teaching.

There are many quantum computing and quantum information textbooks, covering the vast and deep aspects of this subject and its intricate interdisciplinary nature. It is easy for a first-time learner to get lost in this ocean of knowledge. So did we when we started learning this beautiful subject. As an attempt to navigate through this ocean, we decided to curate and bring together the vast topics in an easy-to-understand and concise form, leading to the writing of this expository manuscript. Additionally, the Quantum Information and Computing course at our institute was a primary factor that intrigued and motivated us to write this text. As students and budding researchers ourselves, we understand the struggles and natural questions that arise while learning this subject. In this text, we have made our best attempts to clarify these subtle details and give a flavour of this interdisciplinary subject to anyone, irrespective of their background, provided the reader is comfortable with elementary linear algebra and high school mathematics. This has been a series of consistent efforts in bringing out an extremely comprehensive review of necessary and relevant topics in the broad field of quantum computing and quantum information, in order to provide a head start to early-career researchers, and to attempt to tackle foundational problems in the field.

This text requires dedicated reading and is not meant as a casual introduction to quantum computing and information. It is ideally suited for undergraduates, graduates, and probably motivated high schoolers who want to gain an in-depth yet first-level exposure to the broad field of quantum computing and quantum information theory. This text does not elaborate on experimental aspects and technological developments of the field. We hope that after reading this text, one will be able to delve deeper into any particular topic in quantum computing and quantum information.

Part I gives an overview of how truly interdisciplinary the field of quantum computing and quantum information is, with topics spanning from pure mathematics, theoretical physics and computer science. The first chapter is intended to be a quick recap of all the mathematical tools required to understand this text. The reader is expected to already know most of these, especially matrix and linear algebra. If not already familiar, there are many amazing and standard resources available to learn these topics. (Refer to Linear algebra done right

by Sheldon Axler, or the lectures by Gilbert Strang).

The second chapter sets the stage and provides the necessary background for readers from mathematics, computer science, engineering or any other background who are not familiar with quantum physics. If the reader is already familiar with undergrad-level abstract algebra and quantum mechanics, then they can skip the first two chapters of Part I. However, reading these chapters may provide a quick revision and also present these concepts through a new lens.

The third chapter provides the necessary theoretical computer science background, covering topics primarily in computational complexity theory. Even if the reader is familiar with complexity theory, we encourage them to take a look at this chapter as it touches upon quantum complexity theory alongside the relatively familiar classical complexity theory.

The final chapter of Part I introduces qubits, the fundamental unit of quantum information, serving as the quantum analogue of the classical bit. It lays out the essential concepts and overarching themes of quantum computing and information from functional and historical perspectives. With this foundation, readers are free to explore any chapter from Parts II and III in any order they prefer, as each chapter is self-contained and independent, rather than hierarchically structured. This makes the text equally suitable for readers interested in gaining a broad overview, focusing on a specific topic within quantum computing and quantum information.

For readers who wish to learn independently and work through the entire textbook, we recommend following the chapters in the order presented.

Part II of the text covers topics in quantum computing from basic quantum algorithms to sophisticated algorithms like Shor's prime factoring algorithm that can break the classically secure RSA cryptosystems and Grover's search algorithm that can search in an unstructured database faster than its classical counterpart.

Part III, the last part, talks about quantum information, including topics from quantum error correction. Both Part II and Part III have a lot of visual elements, presenting each topic in an intuitive and pedagogical form. We have made an effort to naturally build the concepts from the ground up rather than directly presenting them. Throughout these two parts, wherever necessary, we have drawn detailed parallels to classical computer science to appreciate the similarities and differences in both these worlds.

Although we have not included exercises, this text offers a concise yet substantial foundation for anyone seeking to build a strong theoretical understanding in a relatively short read. Readers can use it to acquire the necessary background and then practice with problems from other well-known texts on quantum computing and information. Beyond self-study, this text also serves as an ideal companion for any quantum computing or information course. As part of a course throughout the semester, we managed to cover the various aspects that were taught to us, and presented newer insights and perspectives that we believe would help grasp some of the intricacies and inner understandings that make this field

extremely interesting.

We have further provided additional references through footnotes, leading to research material, for the interested reader. We have built some examples through boxes, where we provide a natural, geometric, and intuitive visualisation of some concepts. Additionally, we ensured to have all illustrations generated through L^AT_EX, to ensure that we can convey information through maximum flexibility, and enhance the readability of the text.

Despite having revised and refined this text multiple times, there is always room for improvement. We welcome your feedback. Please feel free to contact the authors with any suggestions or report any errors you may find. We plan to continue updating this expository manuscript, and the latest version will always be available on the website: <https://o-qcblog.github.io/QIQC/>.

Happy learning!

$$\frac{1}{\sqrt{2}} \left[|\mathcal{N}ishkal \ \mathcal{R}ao\rangle \oplus |\mathcal{S}. \ \mathcal{P}admapriya \rangle \right]$$

Contact Details:

Nishkal Rao: [🌐 https://nishkalrao20.github.io/](https://nishkalrao20.github.io/)
 S. Padmapriya: [🌐 https://padmapriya-s1.github.io/](https://padmapriya-s1.github.io/)
[🌐 https://o-qcblog.github.io/](https://o-qcblog.github.io/)



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Additionally, this work would not have been possible without the wonderful support of our parents, siblings, and friends. Finally, the authors want to thank each other for the incredible fun they had during the journey of writing this text.

Navigation

Among the excellent resources for Quantum Computing and Quantum Information, here are some introductory resources that we have referred to and been inspired by.

- Quantum Computation and Quantum Information, Textbook by Isaac Chuang and Michael Nielsen
A foundational introduction to key concepts in quantum computing, highlighting notable aspects of quantum algorithms, quantum information, and quantum error correction.
- Quantum Information and Computation., Lecture Notes for Physics 229 by John Preskill
A comprehensive and insightful resource for understanding quantum computing and quantum information theory.
- Principles of Quantum Computation And Information; Textbook by Giuliano Benenti, Giuliano Strini, Giulio Casati
Comprehensive two-volume companion designed to enhance understanding of quantum computing through clear pedagogical insights and engaging problems.
- Quantum Computer Science: An Introduction; Textbook by David Mermin
Beautiful introduction to various aspects of quantum computing. Beware of the QBits, though!
- Quantum Computing Since Democritus; Textbook by Scott Aaronson
A philosophical understanding of quantum computing based on complexity, offering valuable insights into physics, mathematics, and theoretical computer science.
- Dancing with Qubits: How Quantum Computing Works and how it Can Change the World; Textbook by Robert S. Sutor
Modern introduction to the concepts in quantum computing and the engineering aspects of the physical theory.
- Quantum Computing, Lecture Notes by Rajat Mittal
A concise theoretical computer science and mathematical perspective on quantum computation targeted at an audience lacking a physics background.

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Conventions

Basic Mathematical Notations

\mathbb{N}	The set of natural numbers $\{1, 2, 3, \dots\}$
\mathbb{Z}	The set of integers $\{\dots, -2, -1, 0, 1, 2, \dots\}$
\mathbb{Q}	The set of rational numbers
\mathbb{R}	The set of real numbers
\mathbb{C}	The set of complex numbers
$[n]$	The set $\{1, 2, \dots, n\}$
ϕ	The empty set
\in, \notin	Element / not an element of a set
\subseteq, \subset	Subset / proper subset
\cup, \cap, \setminus	Union, intersection, set difference
$ S $	Cardinality (size) of set S
$\lceil x \rceil, \lfloor x \rfloor$	Ceiling, floor of real x
$n!$	Factorial of n : $n! = 1 \cdot 2 \cdot \dots \cdot n$
$\binom{n}{k}$	Binomial coefficient, number of k -subsets of an n -set
$\sum_{i=1}^n, \prod_{i=1}^n$	Summation and product
$\lim_{n \rightarrow \infty}$	Limit
$\det(A)$	Determinant of matrix A
$\ x\ _p$	p -norm of vector x (for $p \geq 1$)
δ_{ij}	Kronecker delta: 1 if $i = j$, 0 otherwise
$\mathbf{1}_A$	Indicator function of event/set A ($\mathbf{1}_A(x) = 1$ if $x \in A$, else 0)
\oplus	Direct sum or bitwise XOR (meaning clarified where used)
$ S _2$	Euclidean (or ℓ_2) norm
\Rightarrow, \iff	Implication and equivalence
$\forall, \exists, \exists!$	Universal, existential, unique-existence quantifiers
$\langle a, b \rangle$	Inner product between a and b

Dirac Notation

As mathematicians and physicists work with concepts, we need a concise way of conveying what they mean. Good notation can make a statement or a proof much clearer and more insightful to the reader. Over time, the symbols and expressions that prove to be most useful win out while the others fade away into the archives. In the case of Dirac's bra-ket notation, it has become ubiquitous across quantum mechanics and now quantum computing.

Vectors can come in many flavours, as $\mathbf{v} = (v_1, v_2, \dots, v_n)$ as a tuple (useful for our computer

scientists), equivalently, $v = [v_1 \ v_2 \ \cdots \ v_n]$ as a row vector, $v = \begin{bmatrix} v_1 \\ v_2 \\ \vdots \\ v_n \end{bmatrix}$ as a column vector

(daunting form in physics textbooks). While quantum mechanics is centrally captured by linear algebra, we would be needing extensive use of vectors. To add on to the flavours, let us introduce two more invented by Paul Dirac¹, a theoretical physicist, that we proceed to use extensively further.

Given a vector $\mathbf{v} = (v_1, v_2, \dots, v_n)$, we denote by $\langle v|$, the *bra-v*, is defined as

$$\langle v| = [v_1^* \ v_2^* \ \cdots \ v_n^*]$$

where we take the complex conjugate of each entry. For a vector $\mathbf{w} = (w_1, w_2, \dots, w_m)$, we have the *ket-w*, given by

$$|w\rangle = \begin{bmatrix} w_1 \\ w_2 \\ \vdots \\ w_m \end{bmatrix}$$

as the column vector without conjugations.

Inner Product

When $n = m$ for same dimensions, we can conjunct the notation for the *inner product* of the vectors \mathbf{v} and \mathbf{w} , as

$$\langle v|w\rangle = [v_1^* \ v_2^* \ \cdots \ v_n^*] \begin{bmatrix} w_1 \\ w_2 \\ \vdots \\ w_m \end{bmatrix} = v_1^* w_1 + v_2^* w_2 + \dots v_n^* w_n$$

¹If you read Dirac's *Principles of Quantum Mechanics*, he says to assume the correspondence between a ket and the corresponding bra, which is actually a central area of study called the Riesz Representation Theorem, which Dirac assumed as obvious to the readers. He has no mention of it in his book, and we will respect his legacy by doing the same. That being said, this is the same man who remained completely silent after a student said, "I don't understand the second equation," during a lecture. After being asked why Dirac didn't answer the student's question, Dirac said, "That was not a question, that was a statement." The interested reader can refer to ["Meaning of Riesz representations in a layman's term?" a Math StackExchange post](#) for further insight.

which will be very helpful in further discussions. The norm of a vector denoting its length can be seen, thereby as

$$||\boldsymbol{v}|| = \sqrt{\langle v|v \rangle} = \sqrt{|v_1|^2 + |v_2|^2 + \cdots + |v_n|^2}$$

This is why we have the complex conjugates, so that complex numbers can give the norm.

Outer Product

Further, we will be requiring the notion of the *outer product* wherein we have the operation,

$$|w\rangle\langle v| = \begin{bmatrix} w_1 \\ w_2 \\ \vdots \\ w_m \end{bmatrix} \begin{bmatrix} v_1^* & v_2^* & \cdots & v_n^* \end{bmatrix} = \begin{bmatrix} w_1 v_1^* & w_1 v_2^* & \cdots & w_1 v_n^* \\ w_2 v_1^* & w_2 v_2^* & \cdots & w_2 v_n^* \\ \vdots & \vdots & \ddots & \vdots \\ w_n v_1^* & w_n v_2^* & \cdots & w_n v_n^* \end{bmatrix}$$

Asymptotic Notation

Asymptotic notation is used to compare and understand the behaviour of real-valued functions having a positive integer domain as the input grows large. We will be using it extensively to compare and analyse the scale of algorithms in the classical and quantum settings to help us demarcate the separation and understand the quantum advantage.

Consider two functions $f(x)$ and $g(x)$ that map positive integers to positive real numbers. Then the asymptotic notations are defined as follows:

Big \mathcal{O} -notation

$f(x)$ is said to be big-O of $g(x)$, denoted as $\mathcal{O}(g(x))$, if there exists a constant $c > 0$ and a positive integer constant x_0 such that,

$$\begin{aligned} f(x) &\leq cg(x) \quad \forall x \geq x_0 \\ \implies f(x) &\in \mathcal{O}(g(x)) \end{aligned}$$

As an abuse of notation, it is a common practice to write this as $f(x) = \mathcal{O}(g(x))$.

When $g(x)$ is non-zero, the above statement is equivalent to,

$$\begin{aligned} \lim_{n \rightarrow \infty} \sup \frac{f(x)}{g(x)} &< \infty \\ \implies f(x) &\in \mathcal{O}(g(x)) \end{aligned}$$

Big Ω -notation

$f(x)$ is said to be big-Omega of $g(x)$, denoted as $\Omega(g(x))$ if there exists a constant $c > 0$ and there exists a positive integer constant x_0 such that,

$$\begin{aligned} f(x) &\geq cg(x) \quad \forall x \geq x_0 \\ \implies f(x) &\in \Omega(g(x)) \end{aligned}$$

Also denoted as $f(x) = \Omega(g(x))$.

When $g(x)$ is non-zero, the above statement is equivalent to,

$$\begin{aligned} \lim_{n \rightarrow \infty} \inf \frac{f(x)}{g(x)} &> 0 \\ \implies f(x) &\in \Omega(g(x)) \end{aligned}$$

Θ -notation

Θ -notation gives a tight bound. We say $f(x) = \Theta(g(x))$ when $f(x)$ is both $\Omega(g(x))$ and $\mathcal{O}(g(x))$.

\tilde{O} -notation

\tilde{O} -notation is used to hide the logarithmic factors, that is, if $f(x) = \tilde{O}(g(x))$ implies $f(x) = (\log x)^c g(x)$, where c can be any real number.

Little o-notation

$f(x)$ is said to be little-o of $g(x)$, denoted as $o(g(x))$ if for every positive constant $c > 0$ there exists a positive integer constant x_0 such that,

$$\begin{aligned} f(x) &\leq cg(x) \quad \forall x \geq x_0, c > 0 \\ \implies f(x) &\in o(g(x)) \end{aligned}$$

Also denoted as $f(x) = o(g(x))$. Little-o is used to denote a stronger statement, thus giving a looser upper bound to a function compared to big-O, as the above should hold for every $c > 0$ and not just a particular constant. In other words, $g(x)$ grows much faster than $f(x)$, or $f(x)$ grows much slower than $g(x)$.

When $g(x)$ is non-zero, the above statement is equivalent to,

$$\begin{aligned} \lim_{n \rightarrow \infty} \frac{f(x)}{g(x)} &\rightarrow 0 \\ \implies f(x) &\in o(g(x)) \end{aligned}$$

Little ω -notation

$f(x)$ is said to be little-omega of $g(x)$, denoted as $\omega(g(x))$ if for every positive constant $c > 0$ there exists a positive integer constant x_0 such that,

$$\begin{aligned} f(x) &\geq cg(x) \quad \forall x \geq x_0, c > 0 \\ \implies f(x) &\in \omega(g(x)) \end{aligned}$$

Also denoted as $f(x) = \omega(g(x))$. Little-omega is used to denote a stronger statement or looser lower bound to a function compared to big-Omega, as the above should hold for every $c > 0$ and not just a particular constant. In other words, $f(x)$ grows much faster than $g(x)$, or $g(x)$ grows much slower than $f(x)$.

When $g(x)$ is non-zero, the above statement is equivalent to,

$$\begin{aligned} \lim_{n \rightarrow \infty} \frac{f(x)}{g(x)} &\rightarrow \infty \\ \implies f(x) &\in \omega(g(x)) \end{aligned}$$

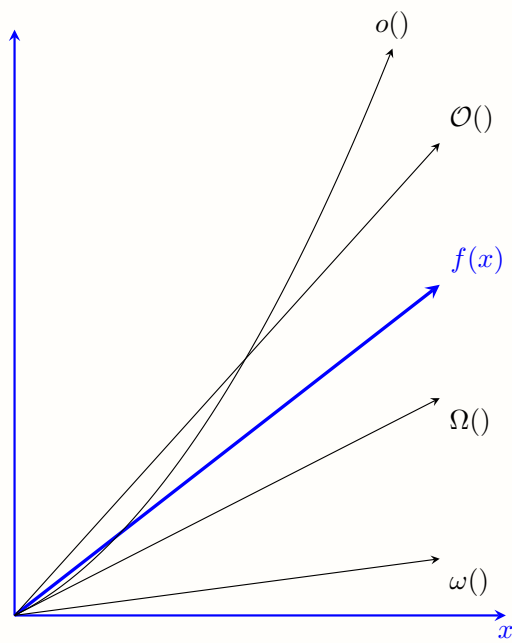


Figure 1: Asymptotic Notation

Part I

Foundations

Chapter 1

Mathematical Background

“If all of mathematics disappeared, physics would be set back by exactly one week.”

– Richard Feynman, *Lecture at Caltech, Pasadena*

“Precisely the week in which God created the world.”

– Mark Kac, *Enigmas of Chance*

1.1 Probability Theory

At its core, quantum mechanics is a probabilistic theory¹, meaning that it predicts the likelihood of different outcomes for a given measurement. The interpretation of these probabilities has been a subject of debate among physicists and philosophers for decades. We provide a concise recap of basic probability theory to address problems in quantum mechanics further:

1. *Conditional probability:* Let A and B be two events

$$P[A|B] := \frac{P(A \cap B)}{P(B)}$$

2. *Partition formula:* Given A and disjoint partition $B_1, B_2 \dots B_m$ of sample space,

$$P(A) = \sum_{i=1}^m P(B_i)P[A|B_i]$$

3. *Bayes rule:*

$$P[A|B] = \frac{P[B|A]P(A)}{P(B)}$$

¹Refer ["Where Quantum Probability Comes From"](#), a [Quantamagazine article](#) for a very interesting insight.

4. *Random variable:* Given sample space Ω of an experiment, a random variable is a function $X : \Omega \rightarrow \mathbb{R}$. In general, the range of a random variable X need not be real; it could be any other set with more structure (like real numbers are *ordered*; they can be added, multiplied, etc.)
5. *Probability mass function:* Given a probability function P on Ω , it can be naturally extended to the probability of the random variable,

$$P_X(x) := P(X = x) = \sum_{w: X(w)=x} P(w)$$

This is the *probability mass function* of a random variable. The *joint probability mass function* is defined to be $P_{X,Y}(x, y) := P(X = x \text{ and } Y = y)$

6. *Expectation* $E[X] := \sum_{x \in R} P(X(w) = x)x$ where R is the range of the random variable X . The expectation is linear, that is $E[aX + bY] = aE[X] + bE[Y]$.
7. *Variance:* $\text{Var}[X] := E[(X - E[X])^2] = E[X^2] - (E[X])^2$. *Standard deviation* is the square root of variance. If $Y = aX$, where X is a random variable, then $\text{Var}[Y] = a^2 \text{Var}[X]$.
8. Let $\{X_i\}_{i=1}^n$ be *pairwise independent family of random variables*. Then,

$$\text{Var}\left[\sum_{i=1}^n X_i\right] = \sum_{i=1}^n \text{Var}[X_i]$$

9. *Inclusion-exclusion principle:*

$$P\left(\bigcup_{i \in [n]} A_i\right) = \sum_{S \subseteq [n], S \neq \emptyset} (-1)^{|S|+1} P\left(\bigcap_{i \in S} A_i\right)$$

10. *Union Bound:* When we have a lot of events, it becomes hard to calculate the probability of their unions using the inclusion-exclusion principle. In these cases, a simple union bound can be used to upper bound the probability of their union.

$$P\left(\bigcup_{i \in [n]} A_i\right) \leq \sum_i P(A_i)$$

1.1.1 Law of large numbers

Let the random experiment be modelled by a random variable X . Suppose the experiment is repeated n times. Denote X_1, X_2, \dots, X_n to be n copies of X (they have the same distribution). We also assume that the family of random variables $\{X_i\}_{i=1}^n$ is pairwise independent (any two random variables are independent).

The intuition is, as n gets bigger, the average value of X_1, X_2, \dots, X_n should be close to $\mathbb{E}[X]$. So, define a new random variable,

$$\bar{X} = \frac{\sum_{i=1}^n X_i}{n}.$$

Hence, \bar{X} is the average of n repetitions of X (as a random variable). By linearity of expectation $E[\bar{X}] = E[X]$.

Theorem 1.1.1. *Weak law of large numbers* Define the random variable $\bar{X} = \frac{\sum_{i=1}^n X_i}{n}$, where each X_i has the same distribution as a random variable X and are pairwise independent. Then,

$$P(|\bar{X} - E[X]| \geq a) \leq \frac{\text{Var}[X]}{na^2}$$

1.2 Linear Algebra

Linear algebra provides the language of quantum mechanics. Its concepts, from vector spaces to eigenvalue decompositions, allow us to rigorously formulate and manipulate the state spaces of quantum systems.

1.2.1 Vector Spaces and Inner Product Spaces

Definition 1.2.1 (Vector Space). A vector space V over a field \mathbb{F} is a set equipped with two operations: vector addition and scalar multiplication. These operations satisfy the following axioms for all $\mathbf{u}, \mathbf{v}, \mathbf{w} \in V$ and all scalars $a, b \in \mathbb{F}$:

1. **Associativity of Addition:** $(\mathbf{u} + \mathbf{v}) + \mathbf{w} = \mathbf{u} + (\mathbf{v} + \mathbf{w})$.
2. **Commutativity of Addition:** $\mathbf{u} + \mathbf{v} = \mathbf{v} + \mathbf{u}$.
3. **Existence of Zero:** There exists a unique zero vector $\mathbf{0} \in V$ such that $\mathbf{u} + \mathbf{0} = \mathbf{u}$.
4. **Existence of Additive Inverses:** For every $\mathbf{u} \in V$, there exists a vector $-\mathbf{u}$ such that $\mathbf{u} + (-\mathbf{u}) = \mathbf{0}$.
5. **Distributivity:** $a(\mathbf{u} + \mathbf{v}) = a\mathbf{u} + a\mathbf{v}$ and $(a + b)\mathbf{u} = a\mathbf{u} + b\mathbf{u}$.
6. **Compatibility:** $a(b\mathbf{u}) = (ab)\mathbf{u}$.
7. **Identity:** $1\mathbf{u} = \mathbf{u}$, where 1 is the multiplicative identity in \mathbb{F} .

Definition 1.2.2 (Inner Product Space). An inner product space is a vector space V endowed with an inner product $\langle \cdot, \cdot \rangle : V \times V \rightarrow \mathbb{F}$ satisfying:

1. **Conjugate Symmetry:** $\langle \mathbf{u}, \mathbf{v} \rangle = \overline{\langle \mathbf{v}, \mathbf{u} \rangle}$.
2. **Linearity in the First Argument:** $\langle a\mathbf{u} + b\mathbf{v}, \mathbf{w} \rangle = a\langle \mathbf{u}, \mathbf{w} \rangle + b\langle \mathbf{v}, \mathbf{w} \rangle$.
3. **Positive-Definiteness:** $\langle \mathbf{u}, \mathbf{u} \rangle \geq 0$, with equality if and only if $\mathbf{u} = \mathbf{0}$.

1.2.2 Linear Operators and Spectral Decomposition

The study of linear operators, particularly those that act on finite-dimensional inner product spaces (Hilbert spaces), reveals the structure behind quantum evolution. In quantum mechanics, operators such as the Hamiltonian or measurement observables are Hermitian, ensuring real eigenvalues and a well-behaved spectral decomposition.

Definition 1.2.3 (Linear Operator). *A mapping $\mathbf{A} : V \rightarrow V$ is called a linear operator if for all $\mathbf{u}, \mathbf{v} \in V$ and scalars $c \in \mathbb{F}$, we have*

$$\mathbf{A}(c\mathbf{u} + \mathbf{v}) = c\mathbf{A}(\mathbf{u}) + \mathbf{A}(\mathbf{v}).$$

Definition 1.2.4 (Hermitian Operator). *A Hermitian operator is a linear operator that is self-adjoint, that is, \mathbf{A} is Hermitian when $\mathbf{A} = \mathbf{A}^\dagger$. Where \mathbf{A}^\dagger is the conjugate transpose of \mathbf{A} .*

Theorem 1.2.1 (Spectral Theorem for Hermitian Operators). *Let \mathbf{A} be a Hermitian operator acting on a finite-dimensional inner product space. Then there exists an orthonormal basis of V consisting of eigenvectors of \mathbf{A} , and the operator can be expressed as*

$$\mathbf{A} = \sum_i \lambda_i \mathbb{P}_i,$$

where $\lambda_i \in \mathbb{R}$ are the eigenvalues and \mathbb{P}_i are the orthogonal projection operators onto the corresponding eigenspaces.

Example 1.2.1. *Consider the operator*

$$\mathbf{A} = \begin{pmatrix} 2 & 1 \\ 1 & 2 \end{pmatrix}$$

acting on \mathbb{R}^2 . A straightforward calculation shows that its eigenvalues are $\lambda_1 = 3$ and $\lambda_2 = 1$, with corresponding normalized eigenvectors. The spectral decomposition then takes the form

$$\mathbf{A} = 3\mathbb{P}_1 + 1\mathbb{P}_2,$$

which reveals the underlying structure of \mathbf{A} in a clear and elegant way.

The spectral theorem tells us that every Hermitian operator can be "diagonalized" by choosing an appropriate basis. This is analogous to expressing a musical chord as a combination of pure tones, with each eigenvalue representing a "note" of the operator, and the corresponding eigenvectors provide the "directions" in the space along which these notes resonate.

1.2.3 Singular Value Decomposition (SVD)

The Singular Value Decomposition (SVD) is a powerful factorization method that generalizes the spectral decomposition to any (possibly non-square) matrix. In the context of quantum computing, SVD is instrumental in understanding state transformations and noise processes.

Theorem 1.2.2 (Singular Value Decomposition). *For any $m \times n$ matrix \mathbf{M} , there exist unitary matrices \mathbf{U} (of size $m \times m$) and \mathbf{V} (of size $n \times n$) such that*

$$\mathbf{M} = \mathbf{U}\mathbf{\Sigma}\mathbf{V}^\dagger,$$

where $\mathbf{\Sigma}$ is an $m \times n$ diagonal matrix with non-negative real numbers (the singular values) on the diagonal.

Example 1.2.2. *Let's look at the matrix*

$$\mathbf{M} = \begin{pmatrix} 1 & 0 \\ 0 & 2 \\ 0 & 0 \end{pmatrix}$$

This matrix takes a 2D vector (since it has 2 columns) and transforms it into a 3D vector (since it has 3 rows). The SVD of this matrix is

$$\mathbf{M} = \underbrace{\begin{pmatrix} 0 & 1 & 0 \\ 1 & 0 & 0 \\ 0 & 0 & 1 \end{pmatrix}}_{\mathbf{U}} \underbrace{\begin{pmatrix} 2 & 0 \\ 0 & 1 \\ 0 & 0 \end{pmatrix}}_{\mathbf{\Sigma}} \underbrace{\begin{pmatrix} 0 & 1 \\ 1 & 0 \end{pmatrix}^\dagger}_{\mathbf{V}^\dagger}$$

Let's understand the components: In this specific case, \mathbf{V} is the matrix that swaps the standard basis vectors.

$$\mathbf{V} = \begin{pmatrix} 0 & 1 \\ 1 & 0 \end{pmatrix}$$

This is a simple rotation and reflection matrix. Since it's a real matrix, the dagger operation is just the transpose, so $\mathbf{V}^\dagger = \mathbf{V}$. When we apply \mathbf{V}^\dagger to an input vector, it swaps its components. For example, it rotates the vector $\begin{pmatrix} 1 \\ 0 \end{pmatrix}$ to $\begin{pmatrix} 0 \\ 1 \end{pmatrix}$.

\mathbf{U} is a 3×3 unitary matrix.

$$\mathbf{U} = \begin{pmatrix} 0 & 1 & 0 \\ 1 & 0 & 0 \\ 0 & 0 & 1 \end{pmatrix}$$

This matrix also performs a rotation in 3D space. Specifically, it swaps the first two coordinates of a vector, which corresponds to a rotation in the output space.

The SVD reveals the core action of \mathbf{M} . The transformation first swaps the input components (\mathbf{V}^\dagger), then stretches the new second component by 2 and the new first component by 1 ($\mathbf{\Sigma}$), and finally swaps these first two components in the 3D output space (\mathbf{U}). The magic of SVD is that it finds the exact "input" and "output" bases (\mathbf{V} and \mathbf{U}) where the transformation is just a simple scaling ($\mathbf{\Sigma}$).

SVD can be seen as a "best possible" diagonalization of any matrix. Imagine reshaping an arbitrary transformation into a rotation (via \mathbf{V}^\dagger), followed by a scaling (via $\mathbf{\Sigma}$), and then another rotation (via \mathbf{U}). This perspective is particularly useful in quantum information, where one often needs to analyze the effect of noise or perform optimal approximations of unitary evolutions.

1.2.4 Polar Decomposition

Think of any quantum process, from a perfect, idealized gate to a noisy, real-world interaction, as a linear transformation acting on a quantum state. The Polar Decomposition provides an essential and deeply intuitive way to dissect any such transformation.

It elegantly separates the process into two fundamental and distinct actions: a pure rotation (represented by the unitary operator \mathbf{U}), which preserves the geometric relationships within the quantum state space, and a pure stretch or deformation (represented by the positive operator \mathbf{J}). This separation is invaluable in quantum information because it allows us to isolate the ideal, coherent part of an evolution from its non-unitary components, which often correspond to noise or measurement effects. Its most vital application is in finding the closest ideal quantum gate (\mathbf{U}) to an actual, imperfect experimental operation, making it an indispensable tool for analyzing gate fidelity and designing error-resilient quantum controls.

Polar decomposition says that every linear operator can be decomposed as a unitary and a unique positive operator.

Theorem 1.2.3 (Polar Decomposition). *Given a linear operator A on a vector space V , there exists unitary U and unique positive matrices $J \equiv \sqrt{A^\dagger A}$ and $K \equiv \sqrt{AA^\dagger}$ such that*

$$A = UJ = KU$$

If A is invertible, then U is also unique.

Proof. J as defined in the theorem is a positive operator, so by spectral decomposition it can be written as $J = \sum_i \lambda_i |i\rangle\langle i|$. Define $|\phi_i\rangle = A|i\rangle$. Notice that $\langle\phi_i|\phi_i\rangle = \lambda_i^2$, so $|\phi_i\rangle/\lambda_i^2$ (when $\lambda_i \neq 0$) is a unit vector, call it $|e_i\rangle$. By the Gram-Schmidt process, we can extend the basis as $\{e_i\}$ to form an orthonormal basis.

Define $U = \sum_i |e_i\rangle\langle i|$. Consider the action of UJ on any eigen vector $|i\rangle$ with non-zero eigenvalue λ_i ,

$$UJ|i\rangle = \lambda_i U|i\rangle = \lambda_i |e_i\rangle = |\phi_i\rangle = A|i\rangle$$

For $\lambda_i = 0$,

$$UJ|i\rangle = 0 = |\phi_i\rangle$$

Thus, the action of UJ on the basis vectors is the same as that of A . So we have $A = UJ$.

$A^\dagger = JU^\dagger$, so $A^\dagger A = J^2$, giving J a unique value $\sqrt{A^\dagger A}$. When A is invertible J also is invertible, thus giving $U = AJ^{-1}$ uniquely. Similar arguments can be used to show $A = KU$ as well. ■

1.2.5 Positive Semidefinite Matrix

Definition 1.2.5 (Positive Semidefinite Matrix). *A $n \times n$ matrix M is called positive semidefinite if $\forall x \in \mathbb{R}^n, x^T M x \geq 0$.*

Theorem 1.2.4. *The following are equivalent,*

1. M is positive semidefinite
2. All eigenvalues of M are non-negative
3. $\exists B$ a $m \times n$ matrix ($m \leq n$) such that $M = B^T B$

1.3 Group Theory

Group theory provides the language of symmetry that is pervasive in both classical and quantum systems. Its axioms encapsulate the essence of symmetry operations, which are central to understanding quantum dynamics and computational processes.

1.3.1 Fundamental Definitions

Definition 1.3.1 (Group). A group $(G, *)$ is a set G together with a binary operation $*$: $G \times G \rightarrow G$ satisfying:

1. **Closure:** For every $a, b \in G$, the product $a * b$ is in G .
2. **Associativity:** For all $a, b, c \in G$, $(a * b) * c = a * (b * c)$.
3. **Identity:** There exists an element $e \in G$ such that for every $a \in G$, $e * a = a * e = a$.
4. **Inverses:** For each $a \in G$, there exists an element $a^{-1} \in G$ such that $a * a^{-1} = a^{-1} * a = e$.

Definition 1.3.2 (Abelian Group). A group G is called abelian (or commutative) if, in addition to the group axioms, it satisfies

$$a * b = b * a \quad \text{for all } a, b \in G.$$

Remarks. Generally, as a shorthand, the group operation symbol is dropped. That is, $g * h$ is written as gh .

1.3.2 Subgroups, Cosets, and Normality

Understanding substructures within a group allows us to analyze and decompose complex symmetry operations.

Definition 1.3.3 (Subgroup). A non-empty subset $H \subseteq G$ is a subgroup of G if H is itself a group under the operation inherited from G . We denote this by $H \leq G$.

Theorem 1.3.1 (Lagrange's Theorem). If G is a finite group and H is a subgroup of G , then the order (number of elements) of H divides the order of G .

Definition 1.3.4 (Normal Subgroup). A subgroup $N \leq G$ is normal (denoted $N \triangleleft G$) if it is invariant under conjugation; that is, for every $n \in N$ and every $g \in G$, we have

$$gng^{-1} \in N.$$

Normal subgroups allow the construction of quotient groups, which capture the idea of symmetry modulo some invariant structure.

1.3.3 Cyclic Groups and Group Homomorphisms

Cyclic groups are the simplest examples of groups, serving as building blocks for more intricate symmetry operations.

Definition 1.3.5 (Cyclic Group). *A group G is cyclic if there exists an element $g \in G$ such that every element in G can be written as a power of g , i.e.,*

$$G = \{g^n \mid n \in \mathbb{Z}\}.$$

Such an element g is called a generator of G .

Example 1.3.1. *The group $(\mathbb{Z}/n\mathbb{Z}, +)$ is cyclic, with 1 (or any element coprime to n) serving as a generator. This example illustrates how modular arithmetic naturally leads to cyclic group structures.*

A deeper understanding of group structure is achieved via homomorphisms.

Definition 1.3.6 (Group Homomorphism). *A map $\varphi : G \rightarrow H$ between two groups (G, \cdot) and $(H, *)$ is a group homomorphism if for all $a, b \in G$,*

$$\varphi(a \cdot b) = \varphi(a) * \varphi(b).$$

Homomorphisms preserve the algebraic structure, allowing us to relate different groups through their shared symmetries.

1.4 Fourier Transformation

Fourier transformation is a fundamental tool that decomposes functions into their constituent frequency components. In both classical and quantum contexts, it enables us to switch between time (or spatial) representations and frequency domains. This dual perspective is not only mathematically elegant but also pivotal in quantum algorithms such as Shor's algorithm.

For a function $f : \mathbb{R} \rightarrow \mathbb{C}$, the Fourier transform is defined as

$$\hat{f}(\xi) = \int_{-\infty}^{\infty} f(x) e^{-2\pi i x \xi} dx.$$

The inverse Fourier transform recovers the original function:

$$f(x) = \int_{-\infty}^{\infty} \hat{f}(\xi) e^{2\pi i x \xi} d\xi.$$

Think of the Fourier transform as a way to “listen” to the hidden frequencies within a signal. Just as a musical chord can be decomposed into individual notes, any function can be expressed as a sum (or integral) of sinusoidal components. In quantum mechanics, this idea underpins the relationship between position and momentum representations.

1.5 Group Theoretic Perspective on Fourier Transform

The Fourier transform can be generalized to functions defined on groups, revealing deep connections between harmonic analysis and group theory. This perspective is especially fruitful in quantum computing, where symmetries play a central role in algorithm design.

1.5.1 Fourier Transform over Abelian Groups

For a finite Abelian group G , the Fourier transform decomposes a function $f : G \rightarrow \mathbb{C}$ into a sum over the group's characters. A *character* χ is a homomorphism from G to the multiplicative circle group \mathbb{C}^\times .

Definition 1.5.1 (Fourier Transform on Finite Abelian Groups). *Let G be a finite Abelian group of order $|G|$. For a function $f : G \rightarrow \mathbb{C}$, its Fourier transform is defined as*

$$\hat{f}(\chi) = \frac{1}{\sqrt{|G|}} \sum_{g \in G} f(g) \overline{\chi(g)},$$

for every character χ in the dual group \hat{G} .

The inverse Fourier transform is given by

$$f(g) = \frac{1}{\sqrt{|G|}} \sum_{\chi \in \hat{G}} \hat{f}(\chi) \chi(g).$$

In the Abelian case, the characters serve as the “frequency modes” of the group. They allow us to express a function as a combination of these basic oscillatory components. For example, in the cyclic group \mathbb{Z}_n , the characters are simple exponential functions, which makes the discrete Fourier transform a natural tool for digital signal processing and quantum algorithms.

Example 1.5.1. *For the cyclic group $G = \mathbb{Z}_n$, the characters are given by*

$$\chi_k(j) = e^{2\pi i k j / n}, \quad k, j \in \{0, 1, \dots, n-1\}.$$

Thus, the Fourier transform on \mathbb{Z}_n becomes

$$\hat{f}(k) = \frac{1}{\sqrt{n}} \sum_{j=0}^{n-1} f(j) e^{-2\pi i k j / n}.$$

1.5.2 Fourier Transform over Non-Abelian Groups

When the group G is non-Abelian, the Fourier transform is extended by replacing characters with the set of irreducible representations. For a finite non-Abelian group G , let $\{\rho\}$ denote the set of inequivalent irreducible representations of G , where each representation $\rho : G \rightarrow \text{GL}(V_\rho)$ maps group elements to matrices acting on the vector space V_ρ .

Definition 1.5.2 (Fourier Transform on Finite Non-Abelian Groups). *Let $f : G \rightarrow \mathbb{C}$ be a function. The Fourier transform of f at an irreducible representation ρ is defined by*

$$\hat{f}(\rho) = \sum_{g \in G} f(g) \rho(g)^\dagger.$$

Here, $\hat{f}(\rho)$ is a matrix of dimension $\dim(\rho) \times \dim(\rho)$.

The inversion formula is given by

$$f(g) = \frac{1}{|G|} \sum_{\rho} \dim(\rho) \operatorname{Tr}(\rho(g) \hat{f}(\rho)),$$

where the sum is taken over all inequivalent irreducible representations of G .

In non-Abelian groups, the irreducible representations generalize the notion of frequency modes. Instead of scalar oscillations, the decomposition yields matrix-valued components that capture more complex symmetries. This richer structure is central in quantum algorithms that exploit non-Abelian hidden subgroup problems or study symmetry properties of quantum systems.

Example 1.5.2. *Consider the symmetric group S_3 , one of the simplest non-Abelian groups. It has three irreducible representations: two one-dimensional representations and one two-dimensional representation. When applying the Fourier transform to a function on S_3 , the one-dimensional representations yield scalar components, while the two-dimensional representation provides a 2×2 matrix capturing the more intricate symmetry of the group.*

1.6 Number Theoretic Foundations

The elegant machinery and beauty of number theory, particularly the properties of modular arithmetic, provide the foundational framework for powerful algorithms, notably in cryptography and quantum computation. We will be using the following beautiful definitions and theorems in the chapters ahead, with a special emphasis on Shor's breakthrough in quantum computing.

1.6.1 Finite Groups Modulo N

Let $(\mathbb{Z}/n\mathbb{Z})^\times$ denote the multiplicative group of integers modulo N that are coprime to N . Formally $(\mathbb{Z}/n\mathbb{Z})^\times = \{a \in \mathbb{Z} \mid 1 \leq a < N \text{ and } \gcd(a, N) = 1\}$.

If n is not a prime, it has all elements of $(\mathbb{Z}/n\mathbb{Z})$ which are coprime with n . If n is prime, $(\mathbb{Z}/n\mathbb{Z})^\times$ is the same as $(\mathbb{Z}/n\mathbb{Z})$. This set forms a group under multiplication modulo N , with order $\varphi(N)$, where φ is Euler's totient function. This denotes the cardinality of numbers, which are coprime to N .

Example 1.6.1. *For $N = 15$, $(\mathbb{Z}/15\mathbb{Z})^\times = \{1, 2, 4, 7, 8, 11, 13, 14\}$ and $\varphi(15) = 8$.*

Note that the Euler totient function is a multiplicative function, that is, if two numbers m and n are relatively prime, that $\varphi(mn) = \varphi(m)\varphi(n)$.

Example 1.6.2. *For $N = 3$, $(\mathbb{Z}/3\mathbb{Z})^\times = \{1, 2\}$ and for $(\mathbb{Z}/5\mathbb{Z})^\times = \{1, 2, 3, 4\}$. Note that $\varphi(3) \times \varphi(5) = \varphi(15) = 8$.*

1.6.2 Order of an Element

Definition 1.6.1 (Order). The **order** of an element $a \in (\mathbb{Z}/n\mathbb{Z})^\times$ is the smallest positive integer r such that: $a^r \equiv 1 \pmod{N}$.

By Lagrange's theorem, we emphasize that the order of the cyclic group generated by an element a given by $\{1, a, a^2, \dots, a^{r-1}\}$ such that $a^r = 1 \pmod{N}$, divides the order of the group, hence r divides $\varphi(N)$.

1.6.3 Fermat-Euler Theorem

Theorem 1.6.1. (Fermat's Little Theorem): For any $a \in \mathbb{Z}$, p some prime,

$$a^p \equiv a \pmod{p}.$$

Theorem 1.6.2. (Fermat-Euler): For any $a \in (\mathbb{Z}/n\mathbb{Z})^\times$,

$$a^{\varphi(N)} \equiv 1 \pmod{N}.$$

Euler's theorem is a generalization of Fermat's little theorem (For any prime $\varphi(p) = p - 1$ since there exist $p - 1$ numbers co-prime to p smaller than p , hence $a^{\varphi(p)} \pmod{p} \equiv a^{p-1} \pmod{p} \equiv 1 \pmod{p}$, thereby $a^p \equiv a \pmod{p}$), which can be understood from group theoretic principles. Since the order of any element in $(\mathbb{Z}/n\mathbb{Z})^\times$ divides the order of $(\mathbb{Z}/n\mathbb{Z})^\times$, we have $a^r \equiv 1 \pmod{N}$ where r divides $\varphi(N)$. Thereby we have $\varphi(N) = rk$ for some k , then $a^{\varphi(N)} \pmod{N} \equiv (a^r)^k \pmod{N} = 1 \pmod{N}$.

1.7 Linear and Semidefinite Programming

Linear programming is a type of optimisation problem that has been extensively studied. Optimising is to maximise or minimise a given *objective function* under given *constraints*. In linear programming, the objective function and the constraints are linear functions. One example of linear programming is given below:

$$\begin{aligned} &\text{maximize} && 3x_1 + 2x_2 \\ &\text{subject to} && x_1 + x_2 \leq 4, \\ & && x_1 \leq 2, \\ & && x_2 \leq 3, \\ & && x_1, x_2 \geq 0. \end{aligned}$$

When there is an additional constraint on the variables that demands they be integers, it is called *integer linear programming (ILP)*. It is a well-known fact that integer linear programming is computationally intractable ².

²Integer linear programming is an NP-complete problem. The paper [Papadimitriou \[1981\]](#) provides a simple and elegant proof of the same. Also, one can find a standard reduction from ILP to 3-SAT given in most algorithms or complexity theory textbooks.

Unfortunately, many day-to-day optimisation and combinatorial problems have a natural integer linear programming formalism. To solve the problem efficiently, we give up on the hope of finding the exact solution and remain satisfied with an approximate solution. We do this by dropping the integer constraint, and this is called linear programming relaxation. There are multiple efficient algorithms to now solve our linear program ³.

Semidefinite programming (SDP) is like an elder and stronger cousin of linear programming. It belongs to a more general type of optimising problem, replacing the single-valued variables in the linear programming setup with matrices. Below is an example of a semidefinite program,

$$\begin{array}{ll} \text{maximize} & \langle C, X \rangle \quad \text{over } X \in \mathbb{R}^{2 \times 2}, X \succeq 0 \\ \text{subject to} & \langle I, X \rangle = 1, \end{array} \quad \text{where } C = \begin{bmatrix} 0 & 1 \\ 1 & 0 \end{bmatrix}, \quad I = \begin{bmatrix} 1 & 0 \\ 0 & 1 \end{bmatrix}.$$

Most of the time, one can also give a *semidefinite programming relaxation* to an integer linear program. This can lead to a "better" approximate solution than the one given by linear programming relaxation. In the later chapter 8, we will see an example of one such problem (MAX CUT PROBLEM) for which this is the case. The good news here is that there are multiple algorithms to solve semidefinite programs efficiently ⁴!

³Like simplex and interior point algorithms etc. Read more in [linear programming Wikipedia](#).

⁴Much like linear programming, there are similar techniques to solve SDPs. Refer to [semidefinite programming Wikipedia for more details](#).

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Chapter 2

Physics Formalism

“When searching for harmony in life, one must never forget that in the drama of existence, we are ourselves both actors and spectators.”

– Niels Bohr, *Discussions with Einstein*

2.1 Postulates of Quantum Mechanics

Quantum mechanics, at its heart, is a framework for predicting the behaviour of the universe at its smallest scales. While its implications can seem *bizarre*, the theory itself rests on a few fundamental postulates. These postulates provide the language and machinery for describing physical reality. Instead of merely stating them, let’s explore their physical meaning.

Postulate 1. *States of a quantum system are associated with a unit vector in Hilbert space.*

Everything we can possibly know about an isolated quantum system is encoded in a single mathematical object: a state vector, denoted $|\psi\rangle$. This vector lives in a special complex vector space called a Hilbert space. Because probabilities must sum to one, this state vector is always a unit vector ($\langle\psi|\psi\rangle = 1$). The power of this postulate is the *principle of superposition*, if $|\psi_1\rangle$ and $|\psi_2\rangle$ are valid states, then so is their linear combination, $\alpha|\psi_1\rangle + \beta|\psi_2\rangle$.

Postulate 2. *Observables are associated with Hermitian operators on the system’s Hilbert space.*

Every measurable property of a system, like position, momentum, or spin, is represented by a Hermitian operator acting on that system’s Hilbert space. The necessity for Hermitian operators which are self adjoint, is because the outcome of any real-world measurement must be a real number, and Hermitian operators are guaranteed to have real eigenvalues.

Postulate 3. *States transform via unitary operations. The Schrodinger equation governs time evolution.*

When we aren't looking at it, a quantum system evolves in a perfectly smooth, continuous, and deterministic way. This evolution is described by a unitary transformation. A state $|\psi(t_1)\rangle$ evolves to $|\psi(t_2)\rangle = U|\psi(t_1)\rangle$. A unitary transformation is essentially a rotation in the Hilbert space that preserves the length of the state vector, ensuring that probabilities continue to make sense over time.

Postulate 4 (Quantum Measurement). *When observable A is measured on state $|\psi\rangle$, the set of outcomes is the set of eigenvalues of A given by $\{a_i\}$.*

- i) The probability of obtaining outcome a_i is given by $p(a_i) = |\langle\psi|a_i\rangle|^2$*
- ii) The state of the system after measurement collapses to one of the eigenstates $\{|a_i\rangle\}$ of A .*

The smooth deterministic evolution of the system is violently interrupted by the act of measurement. Measurement in quantum mechanics is a probabilistic event which has enormous philosophical notions and is still debated upon. When an observable A is measured, the only possible results are the eigenvalues $\{a_i\}$ of the operator A . Even if the system was in a vast superposition of states, the measurement outcome is restricted to this specific set of values. However, we cannot, in general, predict the outcome with certainty. We can only predict the probability of obtaining a specific outcome a_i . This is given by the square of the projection of the state vector $|\psi\rangle$ onto the corresponding eigenvector $|a_i\rangle$. That is, $p(a_i) = |\langle a_i|\psi\rangle|^2$. The closer the state $|\psi\rangle$ is aligned with an eigenvector $|a_i\rangle$, the more likely that outcome becomes. The measurement doesn't just report a value; it fundamentally alters the system. Immediately after obtaining the outcome a_i , the system's state is no longer $|\psi\rangle$. It abruptly *collapses* to the corresponding eigenvector $|a_i\rangle$. All information about the original superposition is lost, and the system is now defined by the result of the measurement. This is the source of the inherent randomness in the quantum world. The notion of how this happens is heavily debated upon and is termed the Measurement Problem¹.

2.2 State Vector

As stated in the first postulate, the state vector $|\psi\rangle$ is the fundamental carrier of information in quantum mechanics. It is an element of a Hilbert space \mathcal{H} , a complex vector space equipped with an inner product. For a two-level system, which will be of significant interest further, the Hilbert space is two-dimensional, \mathcal{H}_2 . The standard basis for this space is given by the orthonormal vectors, represented by $|0\rangle$ and $|1\rangle$, which we shall delve into further.

A general state of a two-level system is a superposition of these basis states:

$$|\psi\rangle = \alpha|0\rangle + \beta|1\rangle$$

where $\alpha, \beta \in \mathbb{C}$ are complex amplitudes. The normalization condition $\langle\psi|\psi\rangle = 1$ requires that $|\alpha|^2 + |\beta|^2 = 1$. This condition ensures that the probabilities of measuring the system to be in state $|0\rangle$ or $|1\rangle$ sum to unity.

¹Refer to the paper [Tomaz et al. \[2025\]](#) to question the reality we live in, and review through the different notions and interpretations of one of the fundamental postulates of quantum mechanics.

2.3 Entanglement

When we consider systems composed of more than one two-level system, we encounter one of the most profound and counterintuitive features of quantum mechanics *entanglement*. A composite quantum system, say consisting of two subsystems A and B , is described by a state vector in the tensor product of their individual Hilbert spaces, $\mathcal{H}_{AB} = \mathcal{H}_A \otimes \mathcal{H}_B$.

A state of two two-level systems $|\psi\rangle_{AB}$ is called *separable* if it can be written as a tensor product of individual states of the subsystems:

$$|\psi\rangle_{AB} = |\psi\rangle_A \otimes |\psi\rangle_B$$

A separable state implies that the properties of subsystem A are independent of subsystem B .

As we shall see further, a state is *entangled* if it is not separable. The most famous example of an entangled state is the Bell state $|\Phi^+\rangle$:

$$|\Phi^+\rangle = \frac{1}{\sqrt{2}} (|00\rangle + |11\rangle) \equiv \frac{1}{\sqrt{2}} (|0\rangle_A \otimes |0\rangle_B + |1\rangle_A \otimes |1\rangle_B)$$

This state cannot be factored into a simple product of a state for system A and a state for system B . The consequence of this is that the measurement outcomes of the two systems are perfectly correlated, no matter how far apart they are. If someone measures their system (A) and finds it in the state $|0\rangle$, they would instantly know that their friend's system (B) is also in the state $|0\rangle$. This *spooky action at a distance*, as Einstein famously called it, does not allow for faster-than-light communication, but it is a powerful resource for quantum computation and information protocols like teleportation, which we shall see further. The counterintuitive predictions of quantum mechanics have been confirmed through experiments that measured the polarization or spin of entangled particles at separate locations. These measurements statistically violated Bell's inequality, demonstrating that the correlations produced by quantum entanglement cannot be accounted for by local hidden variables, which are properties inherent to the individual particles themselves. However, although entanglement can create statistical correlations between events occurring in widely separated locations, it cannot be used for faster-than-light communication.

2.4 Measurement

Every dynamical observable q , such as position, momentum, angular momentum, etc., are associated with a Hermitian operator Q . Note that eigenvalues $\{q_i\}$ of a Hermitian operator are real and the non-degenerate eigenvectors of a Hermitian operator are orthogonal. The eigenvectors $\{|q_i\rangle\}$ of a Hermitian operator form a complete orthonormal basis with a spectral decomposition:

$$Q = \sum_{q_i} q_i \mathbb{P}_{q_i},$$

where the sum runs up to the dimension of the Hilbert space \mathcal{H}_N , and \mathbb{P}_{q_i} are projectors. For non-degenerate eigenvalues, we have $\mathbb{P}_{q_i} = |q_i\rangle\langle q_i|$. For a degenerate subspace with

$\{|q_i^{(1)}\rangle, |q_i^{(2)}\rangle, \dots, |q_i^{(r)}\rangle\}$, with each member having the same eigenvalue q_i , then the projector is constructed using all of them as:

$$\mathbb{P}_{q_i} = \sum_j |q_i^{(j)}\rangle \langle q_i^{(j)}|.$$

Measuring an observable Q for a quantum system in state $|\psi\rangle$ results in one of the eigenvalues q_i with probability given by the Born rule:

$$p(q_i) = \langle \mathbb{P}_{q_i} \rangle_\psi = \langle \psi | \mathbb{P}_{q_i} | \psi \rangle = |\mathbb{P}_{q_i} | \psi \rangle|^2.$$

Interestingly, some experimentalists are still investigating if this rule is exact or a first-order approximation². Note that this rule corresponds to an intuition of two different concepts. Firstly, we can regard the measurement probability as the expectation value of the projector over the state $|\psi\rangle$ corresponding to an ensemble average. Also, this can be seen as the corresponding probability amplitude of the system to transfer from a state $|\psi\rangle$ to a state proportional to $\mathbb{P}_{q_i} |\psi\rangle$, with the amplitude defined from the inner product between the states as $\langle \psi | (\mathbb{P}_{q_i} |\psi\rangle)$. Thereby, we further claim that the post-measurement state can be thought of as evolution into the state $\mathbb{P}_{q_i} |\psi\rangle$, given formally by the normalised form as:

$$|\psi_{q_i}\rangle = \frac{\mathbb{P}_{q_i} |\psi\rangle}{|\mathbb{P}_{q_i} |\psi\rangle|} = \frac{\mathbb{P}_{q_i} |\psi\rangle}{\sqrt{\langle \psi | \mathbb{P}_{q_i} | \psi \rangle}}.$$

For a non-degenerate operator, we obtain $p(q_i) = |\langle q_i | \psi \rangle|^2$, and $|\psi_{q_i}\rangle = |q_i\rangle$. Note that we will touch upon this in great detail, as this collapse of a state corresponds to the most intricate theories in quantum mechanics. Since we have modeled everything through unitarity so far, this non-unitary operation of measurement has very different characteristics and nuances that we will explore.

2.5 State Vector vs Density Matrix

We formulate the notions of quantum mechanics in a very different notion, which is a profound realisation of the statistical and probabilistic interpretations.

To gain a better understanding, we review the uncertainties in quantum mechanics first. There are two different kinds of uncertainty in quantum mechanics. The first is the intrinsic quantum mechanical uncertainty due to its features of superposition and probabilistic measurements. Knowing the state of the system completely still implies that we can only make probabilistic statements about the outcomes of some experiments. For example, if we have two particles in a 2-state system, then the system may be in an entangled state like

$$|\psi\rangle = \frac{1}{\sqrt{2}} (|0\rangle|1\rangle + |1\rangle|0\rangle).$$

In this case, it is not possible to say with certainty if a measurement of one of the particles will yield the result 0 or 1.

²For example, the work [Sinha et al. \[2010\]](#) explores this direction.

The second is a classical uncertainty in the preparation of the state of the system. For example, perhaps we think there's a 50% chance that the system is in the state ψ given above and a 50% chance that the system is in a different state $|\Phi\rangle$, given by

$$|\Phi\rangle = \frac{1}{\sqrt{2}} (|0\rangle|1\rangle - |1\rangle|0\rangle).$$

This kind of uncertainty cannot be represented nicely using only state vectors in Hilbert space. While we are already struggling with two particles, imagine the sheer amount of technicalities that need to be specified for a huge system in thermodynamic aspects. It is to be noted that this situation arises even in the case of a single particle.

The density matrix formulation, borrowed from Quantum Statistical Mechanics, can encode both kinds of uncertainty. Because the density matrix can handle both kinds of uncertainty, the density matrix generalizes the normal quantum state vector and lets us handle a wider variety of cases.

2.6 Density Matrix Formalism

In quantum mechanics, the state of a system is conventionally described by a state vector in a Hilbert space. However, as discussed before, when we wish to account for uncertainties, whether they come from inherent quantum indeterminacy or from classical probabilistic mixtures, the state vector is no longer sufficient. In such cases, we turn to the density matrix (or density operator) formalism, which elegantly encapsulates both pure and mixed states.

Let us begin with a single qubit in a pure state, represented by the state vector $|\psi\rangle = \begin{pmatrix} a \\ b \end{pmatrix}$, where $a, b \in \mathbb{C}$ satisfy $|a|^2 + |b|^2 = 1$. For a pure state, the density matrix is defined as the projection operator onto the state:

$$\rho = |\psi\rangle\langle\psi|.$$

Writing this out in the computational basis $\{|0\rangle, |1\rangle\}$, we obtain

$$\rho = |a|^2|0\rangle\langle 0| + ab^*|0\rangle\langle 1| + a^*b|1\rangle\langle 0| + |b|^2|1\rangle\langle 1|,$$

or, equivalently, in matrix form,

$$\rho = \begin{pmatrix} |a|^2 & ab^* \\ a^*b & |b|^2 \end{pmatrix}.$$

This provides us a natural way to compute probabilities, through $\langle 0|\rho|0\rangle = |a|^2$ being the probability of state $|0\rangle$, and $\langle 1|\rho|1\rangle = |b|^2$ is the probability of state $|1\rangle$. This provides us with a complete description of the ensemble.

The diagonal entries of ρ represent the probabilities of measuring the qubit in the states $|0\rangle$ and $|1\rangle$, respectively, while the off-diagonal (or coherent) terms capture the phase relationships, i.e., the quantum coherence between these basis states.

Pure states describe systems with complete knowledge of the quantum state. In practice, however, we often encounter situations where the system is in a probabilistic mixture of different states. Such a scenario is described by a *mixed state*. If the system is prepared in the state $|\varphi_i\rangle$ with probability p_i , then the density matrix is given by

$$\rho = \sum_i p_i |\varphi_i\rangle\langle\varphi_i|,$$

with the conditions,

$$p_i \geq 0, \quad \sum_i p_i = 1.$$

The states $\{|\varphi_i\rangle\}$ are (not necessarily orthogonal in a general ensemble, but can be chosen to form an orthonormal set in the spectral decomposition).

Further, we realise the agnostic interpretation of the density matrix as a characterization of the entire system being the first *moment* (expectation value) of the projectors $|\varphi_i\rangle\langle\varphi_i|$ as seen from the above relation as a probability-weighted average. This helps us in a compact encoding of what is physically realizable. Measuring devices capture probabilities of various measurements, as seen through projective measurements, which is encoded completely in the density matrix as it is characterized by the first moment solely.

This formulation is especially useful when we have incomplete information about the system or when the system is entangled with an external environment. One of the strengths of the density matrix approach is its ability to provide a unified description of both classical uncertainty and quantum superposition. The density matrix ρ possesses several important mathematical properties:

- **Hermiticity:** $\rho^\dagger = \rho$, ensuring that its eigenvalues are real. Taking the Hermitian conjugate of ρ , we have

$$\begin{aligned} \rho^\dagger &= \left(\sum_i p_i |\varphi_i\rangle\langle\varphi_i| \right)^\dagger \\ &= \sum_i p_i (|\varphi_i\rangle\langle\varphi_i|)^\dagger \quad (\text{by linearity}) \\ &= \sum_i p_i |\varphi_i\rangle\langle\varphi_i| \quad (\text{since } (|\varphi_i\rangle\langle\varphi_i|)^\dagger = |\varphi_i\rangle\langle\varphi_i|) \\ &= \rho. \end{aligned}$$

Thus, ρ is Hermitian. This property guarantees that all eigenvalues of ρ are real.

- **Positivity:** ρ is a positive semi-definite operator, which implies that $\langle\phi|\rho|\phi\rangle \geq 0$ for any state $|\phi\rangle$. For an arbitrary state $|\phi\rangle$, we compute

$$\langle\phi|\rho|\phi\rangle = \left\langle \phi \left| \sum_i p_i |\varphi_i\rangle\langle\varphi_i| \right| \phi \right\rangle$$

$$\begin{aligned}
 &= \sum_i p_i \langle \phi | \varphi_i \rangle \langle \varphi_i | \phi \rangle \\
 &= \sum_i p_i |\langle \varphi_i | \phi \rangle|^2.
 \end{aligned}$$

Since $p_i \geq 0$ and $|\langle \varphi_i | \phi \rangle|^2 \geq 0$ for all i , it follows that

$$\langle \phi | \rho | \phi \rangle \geq 0.$$

Thus, ρ is a positive semi-definite operator.

- **Unit Trace:** $\text{Tr}(\rho) = 1$, reflecting the total probability. Using the definition of the trace in any complete orthonormal basis $\{|j\rangle\}$, we have

$$\begin{aligned}
 \text{Tr}(\rho) &= \text{Tr} \left(\sum_i p_i |\varphi_i\rangle \langle \varphi_i| \right) \\
 &= \sum_i p_i \text{Tr}(|\varphi_i\rangle \langle \varphi_i|).
 \end{aligned}$$

But for any normalised state $|\varphi_i\rangle$,

$$\text{Tr}(|\varphi_i\rangle \langle \varphi_i|) = \langle \varphi_i | \varphi_i \rangle = 1.$$

Thus,

$$\text{Tr}(\rho) = \sum_i p_i = 1.$$

Moreover, the purity of a state can be quantified by the trace of ρ^2 . Starting from the spectral decomposition, the square of the density matrix is

$$\begin{aligned}
 \rho^2 &= \left(\sum_i p_i |\varphi_i\rangle \langle \varphi_i| \right) \left(\sum_j p_j |\varphi_j\rangle \langle \varphi_j| \right) \\
 &= \sum_{i,j} p_i p_j |\varphi_i\rangle \langle \varphi_i | \varphi_j \rangle \langle \varphi_j|.
 \end{aligned}$$

If the states $\{|\varphi_i\rangle\}$ are chosen as an orthonormal eigenbasis of ρ , then

$$\langle \varphi_i | \varphi_j \rangle = \delta_{ij},$$

and we have,

$$\rho^2 = \sum_i p_i^2 |\varphi_i\rangle \langle \varphi_i|.$$

Taking the trace,

$$\text{Tr}(\rho^2) = \text{Tr} \left(\sum_i p_i^2 |\varphi_i\rangle \langle \varphi_i| \right)$$

$$\begin{aligned}
&= \sum_i p_i^2 \text{Tr}(|\varphi_i\rangle\langle\varphi_i|) \\
&= \sum_i p_i^2.
\end{aligned}$$

Since $\sum_i p_i = 1$ and $0 \leq p_i \leq 1$, by the properties of probabilities (Cauchy-Schwarz Inequality) we have

$$\sum_i p_i^2 \leq \left(\sum_i p_i \right)^2 = 1,$$

with equality if and only if one of the $p_i = 1$ (i.e., for a pure state). Hence,

$$\text{Tr}(\rho^2) \leq 1.$$

For a pure state, $\text{Tr}(\rho^2) = 1$ whereas for a mixed state, $\text{Tr}(\rho^2) < 1$. This criterion provides a clear operational test for distinguishing between pure and mixed states. Note that ρ^2 is not analogous to the second moment, which would require the variance, unlike the aspect of ρ being seen to the first moment, characterizing expectation values.

An important aspect of the density matrix formalism is its role in computing expectation values. Let O be any observable (a Hermitian operator). The expectation value of O in the state ρ is defined as

$$\langle O \rangle = \sum_i p_i \langle \varphi_i | O | \varphi_i \rangle.$$

On the other hand, using the definition of the trace,

$$\begin{aligned}
\text{Tr}(\rho O) &= \text{Tr} \left(\sum_i p_i |\varphi_i\rangle\langle\varphi_i| O \right) \\
&= \sum_i p_i \text{Tr}(|\varphi_i\rangle\langle\varphi_i| O) \\
&= \sum_i p_i \langle \varphi_i | O | \varphi_i \rangle,
\end{aligned}$$

where we used the cyclic property of the trace and the fact that

$$\text{Tr}(|\varphi_i\rangle\langle\varphi_i| O) = \langle \varphi_i | O | \varphi_i \rangle.$$

Thus, we conclude that

$$\langle O \rangle = \text{Tr}(\rho O).$$

2.7 Reduced Density Operator

When dealing with a composite quantum system, such as an entangled pair, we often want to describe the state of just one of its subsystems. The state vector $|\psi\rangle_{AB}$ describes the entire system, but what is the state of subsystem A alone? This question is answered by

the *reduced density operator*.

Given a composite system AB described by the density operator ρ_{AB} , the reduced density operator for subsystem A is obtained by performing a partial trace over subsystem B , denoted Tr_B

$$\rho_A \equiv \text{Tr}_B(\rho_{AB})$$

The partial trace is an operation that traces out the degrees of freedom of subsystem B , leaving an operator that acts only on the Hilbert space of A . If $\{|b_j\rangle\}$ is an orthonormal basis for the Hilbert space of subsystem B , the partial trace is defined as

$$\rho_A = \sum_j \langle b_j | \rho_{AB} | b_j \rangle$$

The resulting operator ρ_A completely describes all possible measurement outcomes for any observable acting solely on subsystem A .

A remarkable feature of entanglement is revealed here. If the composite system AB is in a pure entangled state, the reduced state of its subsystems will be a mixed state. Let's demonstrate this with the Bell state $|\Phi^+\rangle = \frac{1}{\sqrt{2}}(|00\rangle + |11\rangle)$. The density operator for the composite system is

$$\begin{aligned} \rho_{AB} &= |\Phi^+\rangle\langle\Phi^+| \\ &= \frac{1}{2} (|00\rangle + |11\rangle)(\langle 00| + \langle 11|) \\ &= \frac{1}{2} (|00\rangle\langle 00| + |00\rangle\langle 11| + |11\rangle\langle 00| + |11\rangle\langle 11|) \end{aligned}$$

Now, we compute the reduced density operator for subsystem A by tracing over B ,

$$\begin{aligned} \rho_A &= \text{Tr}_B(\rho_{AB}) = \langle 0 |_B \rho_{AB} | 0 \rangle_B + \langle 1 |_B \rho_{AB} | 1 \rangle_B \\ &= \frac{1}{2} (\langle 0 |_B | 00 \rangle \langle 00 | 0 \rangle_B + \langle 1 |_B | 11 \rangle \langle 11 | 1 \rangle_B) \\ &= \frac{1}{2} (|0\rangle_A \langle 0|_A + |1\rangle_A \langle 1|_A) \\ &= \frac{1}{2} \begin{pmatrix} 1 & 0 \\ 0 & 1 \end{pmatrix} = \frac{1}{2} \mathbb{I} \end{aligned}$$

The state of subsystem A is termed the *maximally mixed state*. It contains no information about the state of the system along any measurement axis, since there is an equal probability of measuring 0 or 1 for any measurement basis. The purity of this state is $\text{Tr}(\rho_A^2) = \text{Tr}(\frac{1}{4}\mathbb{I}) = \frac{1}{2}$, confirming it is a mixed state. This demonstrates a fundamental principle that, for an entangled system, information is stored in the correlations between the subsystems, not in the subsystems themselves. By ignoring one part of an entangled system, we lose all the information about the whole.

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Chapter 3

Theory of Computation

“As long as a branch of science offers an abundance of problems, so long is it alive; a lack of problems foreshadows extinction or the cessation of independent development.”

– David Hilbert, *Mathematical Problems*

Suppose we are tasked with a problem to be solved on a real computer. A computer can be realised as a circuit model, with wires carrying information to be processed through a set of logical operations (gates), which allows us to implement any complex calculation. Given the limited resources of memory, time, and energy, we are tasked to find the best possible sequences of a set of rules to solve it, optimising the resources. Welcome to the field of computational complexity!

Computational complexity theory is a broad field that uses various models, *circuit model*, *RAM model*, *query model*, etc, to better understand resource optimization in every possible way. We will be looking at *circuit model* and *RAM model* in this chapter and *query model* in chapter 7.

3.1 Turing Machine

By definition, an *algorithm* is a set of instructions for solving a problem. The *Turing Machine*¹ provides a firm mathematical framework to aid the intuitive understanding of an algorithm. It was introduced as the fundamental ‘universal computer’ containing the essential elements on which any modern computer is based.

The general idea is strongly implied from what a ‘human computer’² would do. Such a

¹Alan Turing was a brilliant and eccentric persona, credited with solving the Halting Problem through very intuitive arguments and the formal structure of the Turing machine. Some believe the ideas of Turing inspired Kurt Gödel to formulate the *Incompleteness Theorems*. For more, refer to the [Wikipedia article on the book Gödel, Escher, Bach by Douglas Hofstadter](#).

²Back in the day, a computer was one who computed (sadly, mostly women). Read the article [The Gendered History of Human Computers by Clive Thompson](#) that sheds light on this matter, comparing it with the current status of women in computer science.

human computer has limited storage capacity for information, but ideally has an unlimited amount of paper for reading and writing operations. Formally, a Turing machine consists of

- A *tape*, which can be infinite and is subdivided. Each cell division constitutes one letter \mathcal{A}_i from the alphabet $\{\mathcal{A}_1, \mathcal{A}_2, \dots, \mathcal{A}_k\}$ or is blank.
- A *control unit* with states referring to the internal configuration $\{s_1, s_2, \dots, s_l, \mathcal{H}\}$, where \mathcal{H} halts and terminates the computation internal state and the symbol currently being read. Since we want this machine to be physically realizable, the number of possible internal states should be finite.
- A *read/write* head which reads and writes a new symbol in the current cell, overwriting whatever symbol is there, moving backwards or forward one cell, and switching to a new state or halt.

Turing's first result is the existence of a universal machine, whose job is to simulate any other machine described via symbols on the tape. Let us briefly emphasize why this is so groundbreaking. Imagine you were given a set of Turing machines. Through some painstaking work and tweaking, you can build a machine that can solve any problem for you, that can play games, that can watch videos, print text etc.

If this wasn't already enough, Turing's fundamental insight on the Halting problem envisioned ideas that are very simple to grasp. The halting problem questions whether a given problem halts or not. Simple, isn't it? We can't run it for ages, because we are limited on time, space and money. Although sounding very simple, this problem can give insights into profound philosophies. Think about any unsolved conjecture in math. Suppose we could test out the condition for every integer or natural number through a program sequentially, such that it halts when the conjecture fails. Then deciding whether that program ever halts is equivalent to deciding the truth of the conjecture.

But since we still have many unsolved conjectures, it gives hope to believe that there exists no program to solve the halting problem. How can we even prove such a thing? Mr. Turing to the rescue! These types of problems are frequently encountered in logic and solved by explicitly constructing a contradiction against the assumption.

Say, we have a program \mathcal{P} that decides whether a given program Ω halts. We try to analyse the internal dynamics of the problem underpinning some contradiction through it. Generate another program \mathcal{R} through \mathcal{P} , such that \mathcal{R} runs forever if Ω halts given its own code as input, or \mathcal{R} halts if Ω runs forever given its own code as input. In an argument inspired by Russel's paradox³, what happens if we feed the program \mathcal{R} itself? For $\Omega \equiv \mathcal{R}$, the program halts if it runs forever, and runs forever if it halts. Beautiful, isn't it? These logical arguments comprise a basis of proof called *reductio ad absurdum*.

A corollary to this that easily falls out is the well-acclaimed Gödel's incompleteness theorem. This is the beauty of logic! Such a profound statement about the boundaries of mathematics and, thereby, life is contained in this beautiful and intricate yet simple argument by Turing.

³Suppose a barber who shaves all men who do not shave themselves. Who shaves the barber? For a detailed insight into this paradox, refer to the [Wikipedia article on Russell's paradox](#).

3.2 Circuit Model of Computation

We proceed forward to building a real computer, through ideas from the previous sections, but by introducing the *bit*, the fundamental unit of classical information. The bit is defined as a two-valued binary variable, typically encoding 0 and 1. A circuit is made of *wires* and *gates*, with each wire carrying one bit of information, and the gates performing logical operations.

Any number $N < 2^n$ can be encoded as a binary sequence of 0's and 1's as:

$$N = \sum_{k=0}^{n-1} a_k 2^k,$$

where the value of each digit $a_k \in \{0, 1\}$. We represent $N \equiv a_{n-1}a_{n-2}\dots a_1a_0$. The supremacy of binary would be to enable voltage-based regulations for storing information. The binary operations are embodied through logical gates, which respect the Boolean algebra.

In any model of computation, we provide a n -bit input and recover a m -bit output, represented through a logical function as:

$$f : \{0, 1\}^n \rightarrow \{0, 1\}^m.$$

The universality of some elementary logical operations is to embed any operation as a series of elementary logical operations. Any function can be constructed from the elementary gates AND, OR, NOT, and FANOUT, constituting the universal set of gates for classical computation. The number of these basic gates used in a particular algorithm determines the *circuit complexity* of the algorithm.

3.3 RAM Model of Computation

The most commonly used model of computation for the analysis of algorithms is the Random-Access Machine RAM model of computation. In this book, when we talk about asymptotic bounds, unless specified otherwise, we talk about the RAM model of computation. Informally⁴, the following describes the RAM model:

- Has a finite memory that is divided into units of w bits.
- We set $w = \log n$ where n is the upper bound on the size of input received or the upper bound on the size of the computation.
- Basic arithmetic (+, -, ×, %, /), logical (NOT, AND, OR) and relational (>, <, =) operators are considered to take one time unit.
- Function call, accessing a memory location, and bitwise operations are also one time step.

⁴For a more mathematically rigorous understanding, refer to [Nelson \[2016\]](#) or watch the video lectures [O'Donnell \[2020\]](#).

- Other complex operations, which are generally composed of the above unit time operations, have correspondingly multiple time steps.

Thus, for any algorithm, the total number of time steps, calculated as mentioned above, determines the *time complexity* of the algorithm. The size of memory used by the algorithm overall is called *space complexity*.

3.4 Bird's Eye View of Complexity Theory

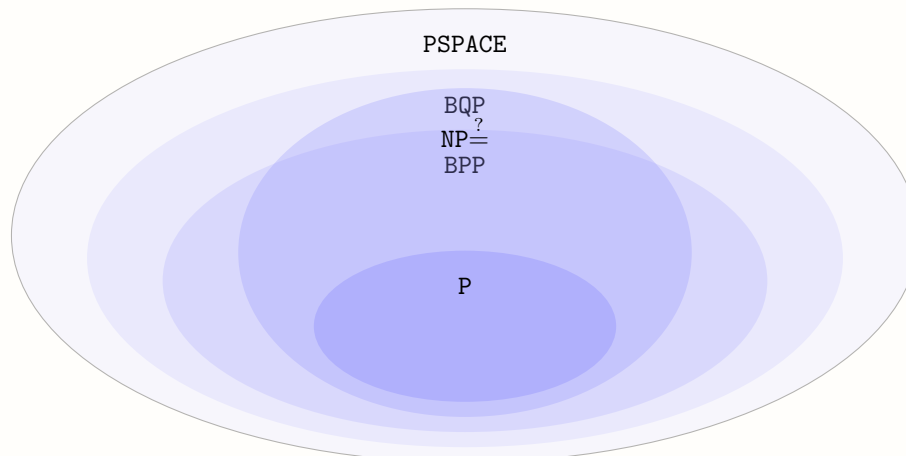
The ability to compute is limited by two resources: space (memory) and time. The difficulty of computing allows problems to be categorized into different complexity classes.

Consider an algorithm that takes in an input of length n (for example, the number of digits in a number). We call this a polynomial time algorithm if it doesn't take more than Cn^k steps for some fixed C , $k \geq 0$, to compute the answer. We denote this by $\mathcal{O}(n^k)$. These are considered efficient algorithms. The class of problems solved by these algorithms is called P .

Another important complexity class is called NP . This is the class of problems whose solutions can be verified in polynomial time. For example, once we find the factorization of some number $N = PQ$, we can efficiently verify that $PQ = N$. Indeed, we have $P \subseteq NP$.

Both complexity classes presented above are bounded by time. There are also a number of complexity classes bounded by space. $PSPACE$ is such a class that contains problems that can be solved with a polynomial number of bits in input size.

For our study, there are two more complexity classes that are important. The first is the BPP , which are problems that can be solved with a bounded probability of error in polynomial time. The second one is the BQP , which is essentially the same thing on a quantum machine. Factoring numbers using Shor's algorithm is BQP .



The known relationship between these complexity classes is:

$$P \subseteq BPP, NP, BQP \subseteq PSPACE$$

In addition, we also have $BPP \subseteq BQP$. The relationship between BPP , NP and BQP is unknown.

3.5 Church-Turing Thesis

Complexity theory classifies problems as *efficient*, which can be solved using resources that are bounded by the size of the input, and *intractable*, which are superpolynomial in the input size. While the former is easy or feasible to solve, the latter is difficult. But how can we solve the tractable ones?

The Church-Turing thesis asserts that any model of computation can be simulated by a Turing machine, with almost a polynomial increase in the number of elementary operations involved. This profound correspondence states that if a problem cannot be solved with polynomial resources on a Turing machine, then we better lose hope!

So far, we have been seeing the story of classical computers. Can quantum computers give hope to solve even the intractable problems? The honest answer is, nobody knows. But over the past few decades, the rise of new quantum algorithms, robust quantum error correction techniques, and the advancement of qubit technology has increased the application of quantum computing to a wide range of disciplines, including machine learning, computational chemistry, biology, quantum simulation of molecules, etc, which has kept quantum computing research positive. It has also led to the introduction of brand new fields like post-quantum cryptography.

Also, it is important to remember that quantum computers won't likely replace classical computers as each of these is specialized in its own way. We will be seeing an example of this in chapter 6, how quantum computing using quantum Fourier transform can solve factoring problem which is a classical NP-hard problem, yet this algorithm can not be used to compute all Fourier coefficients of a function, a polynomial time solvable task on a classical computer. We still do not know what tasks a quantum computer is better at compared to classical computers. All we know is that for certain cherry-picked problems, we certainly have a better algorithm in the quantum world.

Thus, there are a lot of unanswered questions in the field of quantum computing and quantum information, waiting to be solved by future researchers (*intended to the readers*)!

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Chapter 4

Overview of Quantum Computer and Quantum Information

“Computation is Physical. The universe is a [quantum] computer.”

– Seth Lyod, *Programming the Universe*

A brief history¹ of the developments leading to quantum computing and information are presented in the Table 4.1 below. We shall delve into the relevant concepts that we shall be significantly using for quantum computing and information.

4.1 Qubit

The basic unit of information in the classical world is a bit. A bit can be 0 or 1. The quantum world analogue of a bit is a *qubit*. But spooky as quantum mechanics sounds, a qubit can be 0 and 1 at the same time. More precisely, it can be any linear combination of $|0\rangle$ and $|1\rangle$.

Formally, a qubit is a two-level quantum system. It resides in a 2-dimensional complex linear vector space (Hilbert space). With orthonormal basis $\{|0\rangle, |1\rangle\}$. Then the most general normalised state can be expressed as $a|0\rangle + b|1\rangle$ satisfying $|a|^2 + |b|^2 = 1$.

Consider a general qubit $|\psi\rangle$,

$$|\psi\rangle = \alpha|0\rangle + \beta|1\rangle = \alpha \begin{pmatrix} 1 \\ 0 \end{pmatrix} + \beta \begin{pmatrix} 0 \\ 1 \end{pmatrix} = \begin{pmatrix} \alpha \\ \beta \end{pmatrix}$$

¹Inspired from a recent [talk at ICTS by Dr. Ashwin Nayak](#). Abstract: *Although quantum mechanics was established before computability theory, the idea of quantum computers emerged in the 1980s. Initially explored by a select few, quantum computation gained prominence in the 1990s, particularly after the introduction of Shor’s algorithm for integer factorization, which demonstrated the advantages of quantum computing. Since then, the field has evolved, driven by the potential for new technologies and the exploration of innovative ideas.*

1930s	Models of Computation <i>Classes include Recursive functions, λ-calculus, Turing Machine; early formalizations motivated by automating theorem-proving.</i>
1936	Church–Turing thesis <i>Any reasonable model of computation is equivalent in power to the Turing Machine.</i>
1960-70s	Quantum Communication, Cryptography, Quantum Money
1980	Computation is Physical <i>A paradigm shift recognizing that computation must obey the laws of physics; notions further supported by insights from information theory and thermodynamics.</i>
1982	Feynman’s Quantum Computer <i>Feynman proposed using quantum systems to simulate physical processes that are infeasible for classical computers.</i>
1985	Deutsch’s Physical Church–Turing thesis <i>Deutsch extended the Church-Turing thesis to the quantum realm, arguing that a universal quantum computer could efficiently simulate any physical process.</i>
1985-92	Extended Church–Turing thesis in the Classical Context <i>This thesis asserted that any reasonable computational model can be efficiently simulated by a probabilistic Turing machine. Early work by Deutsch and later by Josza, along with insights formalized by Bernstein and Vazirani, began to challenge this view in relativized (oracle) settings.</i>
1993	Bernstein-Vazirani’s Quantum Turing Machine <i>Bernstein and Vazirani rigorously defined the Quantum Turing Machine model and demonstrated, in the oracle (or relativized) setting, a superpolynomial advantage over classical deterministic models.</i>
1994	Simon’s Problem <i>Simon’s algorithm provided the first exponential separation between quantum and classical query complexities, hinting at the power of quantum computation.</i>
1994	Shor’s Algorithm <i>Shor introduced polynomial-time algorithms for integer factorization and discrete logarithms, exploiting periodicity via the Quantum Fourier Transform. This result showcased an exponential advantage over the best-known classical algorithms.</i>
1996	Quantum Parallelism & Formula Evaluation <i>Grover’s algorithm showcased quantum parallelism via superposition, achieving a quadratic speedup for unstructured database search over classical randomized algorithms. This breakthrough not only demonstrated a clear quantum advantage but also inspired new modular techniques in quantum algorithm design.</i>

Table 4.1: Overview of the developments leading to modern-day quantum computation and information.

As $\alpha, \beta \in \mathbb{C}$ we can write them as $z_c = r_c e^{i\theta_c}$

By doing this, we get the *polar representation* of the quantum state:

$$|\psi\rangle = r_\alpha e^{i\theta_\alpha} |0\rangle + r_\beta e^{i\theta_\beta} |1\rangle$$

With some rearrangement and ignoring the overall phase (as a qubit is normalised and in general the overall phase does not matter), we get,

$$\begin{aligned} |\psi'\rangle &= e^{-i\theta_\alpha} (r_\alpha e^{i\theta_\alpha} |0\rangle + r_\beta e^{i\theta_\beta} |1\rangle) \\ &= r_\alpha |0\rangle + r_\beta e^{i(\theta_\beta - \theta_\alpha)} |1\rangle \\ &= r_\alpha |0\rangle + r_\beta e^{i\theta} |1\rangle \end{aligned}$$

$$|\alpha|^2 + |\beta|^2 = 1 \Rightarrow |r_\alpha|^2 + |x + iy|^2 = r_\alpha^2 + x^2 + y^2 = 1$$

This last equation is just a 3-dimensional sphere in real space!

Setting $z = r_\alpha$ we can write the state as,

$$\begin{aligned} |\psi'\rangle &= z|0\rangle + (x + iy)|1\rangle \\ &= \cos \theta |0\rangle + \sin \theta (\cos \phi + i \sin \phi) |1\rangle \\ &= \cos \theta |0\rangle + e^{i\phi} \sin \theta |1\rangle \end{aligned}$$

But notice that the angles are restricted,

$$0 \leq \theta \leq \pi, 0 \leq \phi \leq 2\pi$$

This gives us the general form of a qubit, which can be thought of as a point on a unit sphere specified by the coordinates (θ, ϕ)

$$|\psi\rangle = \cos \frac{\theta}{2} |0\rangle + e^{i\phi} \sin \frac{\theta}{2} |1\rangle$$

This unit sphere on which the qubit lies is called the *Bloch sphere*.

Remarks. The two orthogonal qubits $\{|0\rangle, |1\rangle\}$ are along the z -axis in the Bloch sphere. These two basis kets are called the computational basis states. Later you will learn about the Pauli group and realise that the computational basis is nothing but the eigen basis of the Pauli Z operator or the Phase gate Z .

The state $\frac{|0\rangle + |1\rangle}{\sqrt{2}}$ and $\frac{|0\rangle - |1\rangle}{\sqrt{2}}$ denoted as $|+\rangle$ and $|-\rangle$ respectively are eigen basis of Pauli X operator or the NOT gate X . And the state $\frac{|0\rangle + i|1\rangle}{\sqrt{2}}$ and $\frac{|0\rangle - i|1\rangle}{\sqrt{2}}$ denoted as $|i\rangle$ and $|-i\rangle$ respectively are eigen basis of Pauli Y operator or the Y gate which is equal to $-iXZ$.

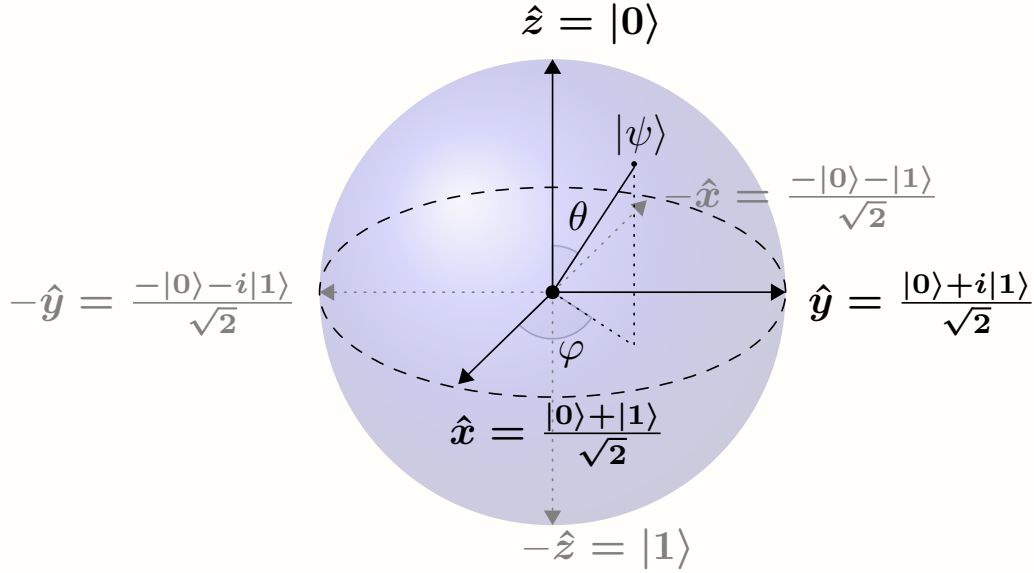


Figure 4.1: Bloch Sphere

4.2 Multi Qubits

Although a single qubit is extremely interesting, but the true power of quantum computation is unleashed when we consider multiple qubits working together. Just as classical computers use registers of many bits, quantum computers use registers of multiple qubits. The way we describe these multi-qubit systems is through a mathematical construction called the *tensor product*.

Let's consider the simplest multi-qubit system of two qubits. If the first qubit is in the state $|\psi_A\rangle = \alpha_0|0\rangle + \alpha_1|1\rangle$ and the second is in the state $|\psi_B\rangle = \beta_0|0\rangle + \beta_1|1\rangle$, the state of the combined two-qubit system is given by their tensor product, denoted $|\psi_A\rangle \otimes |\psi_B\rangle$. The tensor product combines the two vector spaces into a larger one. For a single qubit, the Hilbert space is 2-dimensional (\mathcal{H}_2). For two qubits, the combined Hilbert space $\mathcal{H}_4 = \mathcal{H}_2 \otimes \mathcal{H}_2$ is 4-dimensional.

The tensor product is distributive, so we can expand it as follows:

$$\begin{aligned} |\psi_A\rangle \otimes |\psi_B\rangle &= (\alpha_0|0\rangle + \alpha_1|1\rangle) \otimes (\beta_0|0\rangle + \beta_1|1\rangle) \\ &= \alpha_0\beta_0(|0\rangle \otimes |0\rangle) + \alpha_0\beta_1(|0\rangle \otimes |1\rangle) + \alpha_1\beta_0(|1\rangle \otimes |0\rangle) + \alpha_1\beta_1(|1\rangle \otimes |1\rangle) \end{aligned}$$

For convenience, we use a shorthand notation where $|a\rangle \otimes |b\rangle$ is written as $|ab\rangle$ or $|\psi_A\rangle|\psi_B\rangle$. Using this, the state becomes

$$\alpha_0\beta_0|00\rangle + \alpha_0\beta_1|01\rangle + \alpha_1\beta_0|10\rangle + \alpha_1\beta_1|11\rangle$$

The four states $|00\rangle, |01\rangle, |10\rangle, |11\rangle$ form an orthonormal basis for the two-qubit system. This generalizes powerfully: a system of n qubits is described by a state vector in a 2^n -dimensional Hilbert space. This exponential growth of the state space with the number of qubits is a key reason for the potential power of quantum computers.

It's important to remember that a general n -qubit state is a superposition of all 2^n basis states. For two qubits, an arbitrary state is:

$$|\Psi\rangle = c_{00}|00\rangle + c_{01}|01\rangle + c_{10}|10\rangle + c_{11}|11\rangle$$

where the complex coefficients must satisfy $\sum_{ij} |c_{ij}|^2 = 1$. The above 2^n basis states are also written as numbers 1 to n in base 10, corresponding to the base 2 representation. That is,

$$|\Psi\rangle = c_{00}|0\rangle + c_{01}|1\rangle + c_{10}|2\rangle + c_{11}|3\rangle$$

As we saw in the previous chapter, if a state cannot be written as a simple tensor product of its constituent parts (i.e., it is not separable), it is entangled.

In many quantum algorithms, we also make use of *ancilla qubits*. These are extra helper qubits that are used as a workspace during a computation, much like temporary variables in classical programming. They might be used to store intermediate results or to enable complex controlled operations. Typically, an ancilla qubit is initialized to a known state, like $|0\rangle$, interacts with the primary qubits of the computation through the multi-qubit operation, and is ideally returned to its initial state at the end of the algorithm so it is disentangled from the final result.

4.3 Gates and Circuits

4.3.1 Single Qubit Gates and 2-qubit Gates

As seen in the previous section, a qubit can be thought of as a unit vector in the Bloch sphere. So, what does computation mean in this setup? Given an input qubit, we can think of computation as a series of transformations done to the input qubit to get the desired output qubit state. What type of transformation? Note that the qubit is of unit norm, and norm-preserving transformations are unitary transformations. So these transformations can be captured by unitary matrices.

Some examples of single-qubit gates and two-qubit gates are given below in Fig. 4.2. Notice that all these quantum gates are unitary matrices.²

4.3.2 Constructing Arbitrary 2-qubit States

From the Sec. 4.1, we know that for any $|\psi\rangle$ there is a 1-qubit unitary gate \mathbf{U} that takes $|0\rangle$ to $|\psi\rangle$ such that $\mathbf{U}|0\rangle = |\psi\rangle$.

²For detailed explanation, refer to [Nielsen and Chuang \[2011\]](#).

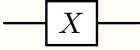
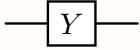

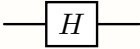
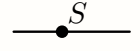
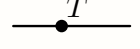
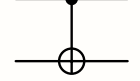
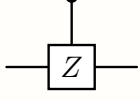
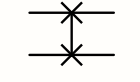
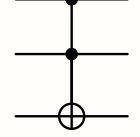
Operator	Gate(s)	Matrix
Pauli-X (X)		$\begin{pmatrix} 0 & 1 \\ 1 & 0 \end{pmatrix}$
Pauli-Y (Y)		$\begin{pmatrix} 0 & -i \\ i & 0 \end{pmatrix}$
Pauli-Z (Z)		$\begin{pmatrix} 1 & 0 \\ 0 & -1 \end{pmatrix}$
Hadamard (H)		$\frac{1}{\sqrt{2}} \begin{pmatrix} 1 & 1 \\ 1 & -1 \end{pmatrix}$
Phase (S, P)		$\begin{pmatrix} 1 & 0 \\ 0 & i \end{pmatrix}$
$\pi/8$ (T)		$\begin{pmatrix} 1 & 0 \\ 0 & e^{i\pi/4} \end{pmatrix}$
Controlled Not (CNOT, CX)		$\begin{pmatrix} 1 & 0 & 0 & 0 \\ 0 & 1 & 0 & 0 \\ 0 & 0 & 0 & 1 \\ 0 & 0 & 1 & 0 \end{pmatrix}$
Controlled Z (CZ)		$\begin{pmatrix} 1 & 0 & 0 & 0 \\ 0 & 1 & 0 & 0 \\ 0 & 0 & 1 & 0 \\ 0 & 0 & 0 & -1 \end{pmatrix}$
SWAP		$\begin{pmatrix} 1 & 0 & 0 & 0 \\ 0 & 0 & 1 & 0 \\ 0 & 1 & 0 & 0 \\ 0 & 0 & 0 & 1 \end{pmatrix}$
Toffoli (CCNOT, CCX, TOFF)		$\begin{pmatrix} 1 & 0 & 0 & 0 & 0 & 0 & 0 & 0 \\ 0 & 1 & 0 & 0 & 0 & 0 & 0 & 0 \\ 0 & 0 & 1 & 0 & 0 & 0 & 0 & 0 \\ 0 & 0 & 0 & 1 & 0 & 0 & 0 & 0 \\ 0 & 0 & 0 & 0 & 1 & 0 & 0 & 0 \\ 0 & 0 & 0 & 0 & 0 & 1 & 0 & 0 \\ 0 & 0 & 0 & 0 & 0 & 0 & 0 & 1 \\ 0 & 0 & 0 & 0 & 0 & 0 & 1 & 0 \end{pmatrix}$

Figure 4.2: Comprehensive list of some of the extensively used single-qubit and two-qubit gates in quantum computation.

An arbitrary 2-qubit state can be written as,

$$|\Psi\rangle = \alpha_{00}|00\rangle + \alpha_{01}|01\rangle + \alpha_{10}|10\rangle + \alpha_{11}|11\rangle.$$

This can be rewritten as

$$|\Psi\rangle = |0\rangle \otimes |\psi\rangle + |1\rangle \otimes |\phi\rangle,$$

where

$$|\psi\rangle = \alpha_{00}|0\rangle + \alpha_{01}|1\rangle, \quad |\phi\rangle = \alpha_{10}|0\rangle + \alpha_{11}|1\rangle.$$

Now consider an unitary $\mathbf{U} \otimes \mathbb{I}$ acting on $|\Psi\rangle$ such that

$$\begin{aligned} (\mathbf{U} \otimes \mathbb{I})|\Psi\rangle &= \mathbf{U}|0\rangle \otimes |\psi\rangle + \mathbf{U}|1\rangle \otimes |\phi\rangle \\ &= (a|0\rangle + b|1\rangle) \otimes |\psi\rangle + (-b^*|0\rangle + a^*|1\rangle) \otimes |\phi\rangle \\ &= |0\rangle |\psi'\rangle + |1\rangle |\phi'\rangle. \end{aligned}$$

where $|\psi'\rangle = a|\psi\rangle - b^*|\phi\rangle$ and $|\phi'\rangle = b|\psi\rangle + a^*|\phi\rangle$.

Note that \mathbf{U} is a unitary that we are constructing. Thus we can pick a and b such that we make $|\psi'\rangle$ and $|\phi'\rangle$ orthogonal. And choose λ and μ such that we make $|\psi''\rangle = \frac{|\psi'\rangle}{\lambda}$, $|\phi''\rangle = \frac{|\phi'\rangle}{\mu}$ unit vectors.

As $|\psi''\rangle, |\phi''\rangle$ are orthonormal, they are related to $|0\rangle$ and $|1\rangle$ by unitary transformation.

$$|\psi''\rangle = V|0\rangle \quad |\phi''\rangle = V|1\rangle.$$

Putting all this together, we can rewrite the equations as

$$\begin{aligned} (\mathbf{U} \otimes \mathbb{I})|\Psi\rangle &= |0\rangle |\psi'\rangle + |1\rangle |\phi'\rangle \\ &= \lambda |0\rangle |\psi''\rangle + \mu |1\rangle |\phi''\rangle \\ (\mathbf{U} \otimes \mathbb{I})|\Psi\rangle &= \mathcal{U}|\Psi\rangle \text{ (say)} \\ \mathcal{U}\mathcal{U}^\dagger |\Psi\rangle &= |\Psi\rangle \text{ (as } \mathcal{U} \text{ is unitary)} \\ \implies |\Psi\rangle &= \mathcal{U}^\dagger (|0\rangle |\psi'\rangle + |1\rangle |\phi'\rangle) = (\mathcal{U}^\dagger \otimes V)(\lambda |00\rangle + \mu |11\rangle) \\ \text{(since } |\psi''\rangle \text{ and } |\phi''\rangle \text{ can be got from } |0\rangle \text{ and } |1\rangle \text{ by an unitary transformation)} \\ &= (\mathcal{U}^\dagger \otimes V)(C_{10}(\lambda |0\rangle + \mu |1\rangle) \otimes |0\rangle) \\ &= (\mathcal{U}^\dagger \otimes V)(C_{10}(W|0\rangle) \otimes |0\rangle) \\ |\Psi\rangle &= \mathcal{U}^\dagger V C_{10} W |00\rangle. \end{aligned}$$

Here C_{10} is a cNOT gate with 1st qubit as the target and the second as the control, and W is yet another unitary. Therefore, three single-qubit unitary and a cNOT gate are enough to get any general 2-qubit state from $|00\rangle$.

4.3.3 Need for n -qubit Gates?

For a general quantum circuit with n qubits, would we need to have all possible n -qubit gates? Fortunately, we need not just keep learning multiple gates for each value of n . It

turns out that arbitrary unitary transformations can be approximated to an arbitrary degree of precision by sufficiently many 1- and 2-qubit gates ³.

Also, the technical challenge in designing higher-order quantum gates is way more challenging than the already existing challenge in making reliable 1 and 2-qubit gates. Thus, the art of designing quantum algorithms lies in carefully picking the unitary transformations that can be built entirely out of products of unitary transformations on 1 and 2 qubits.

Implementation of qubits and gates

As we have seen, qubits are essentially two-level quantum systems. So any two-level quantum system can be used to realise a physical qubit. Currently, multiple such quantum systems are being developed by researchers, each with its own advantages and disadvantages, appropriately used depending on the purpose. Some of them used in practice are *Superconducting*, *Trapped ions*, *Quantum dots*, *Photons* and *Neutral atoms*.

IBM primarily uses superconducting transmon qubits. Superconducting qubits are controlled by microwave pulses. Gates are implemented by carefully tuned microwave pulses (single-qubit rotations) and microwave-activated two-qubit interactions (like cross-resonance), scheduled and calibrated at the pulse level via Qiskit. ^a

^aFor a detailed account, refer to the papers [Kjaergaard et al. \[2020\]](#) and review article [Wendin \[2017\]](#)

4.4 Reversible Computation

As defined in chapter 3, *algorithm* is a set of instructions for solving a problem. Suppose we wish to compute a function f , one can think of the classical computer program as given an input x it performs the necessary instructions and outputs $f(x)$. Similarly we would expect to design a quantum computer to act on x and produce the necessary $f(x)$.

For a function $f : \{0,1\}^n \rightarrow \{0,1\}^m$, a quantum computer needs at least $n + m$ qubits to compute it. Like the Turing Machine, which overwrites on the input tape, why can't we use fewer than $n + m$ qubits to compute f ? One important reason why this can not be done is that if f assigns the same value to different values of x then this computation cannot be inverted if its only effect is to transform the contents of a single register from x to $f(x)$. Thus, the reversibility constraint forces us to have at least $n + m$ qubits to compute f . Thus, computing f is the same as applying a unitary \mathbf{U}_f on the $n + m$ qubits.

\mathbf{U}_f is defined by specifying its action on the basis states. By linearity, this can be extended to any arbitrary superposition of the basis vectors. The standard quantum computation protocol defines the action of \mathbf{U}_f on the computational basis $|x\rangle_n |y\rangle_m$ of the $n + m$ qubits making the *input* and *output* registers as follows:

$$\mathbf{U}_f(|x\rangle_n |y\rangle_m) = |x\rangle_n |y \oplus f(x)\rangle_m$$

³The argument is given with great detail in the paper [Barenco et al. \[1995\]](#)

where \oplus indicates modulo-2 bitwise addition (without carrying) or exclusive OR operation.

If the initial value represented by the output register is $y = 0$ then we have

$$\mathbf{U}_f(|x\rangle_n|0\rangle_m) = |x\rangle_n|f(x)\rangle_m$$

and we end up with $f(x)$ in the output register. Regardless of the initial value of y , the input register remains in its initial state $|x\rangle_n$.

The transformation \mathbf{U}_f is clearly invertible. Also note that \mathbf{U}_f is its own inverse:

$$\begin{aligned}\mathbf{U}_f\mathbf{U}_f(|x\rangle|y\rangle) &= \mathbf{U}_f(|x\rangle|y \oplus f(x)\rangle) \\ &= |x\rangle|y \oplus f(x) \oplus f(x)\rangle = |x\rangle|y\rangle,\end{aligned}$$

since $z \oplus z = 0$ for any z .

Thus, \mathbf{U}_f gives a generic way to construct any function f on a quantum computer.

4.4.1 Information is Physical

Though it appears as if some mental gymnastics have to be done to compute $f(x)$ by first defining \mathbf{U}_f in a quantum computer, this has given an intrinsic advantage to quantum computers over classical irreversible computation.

Be it a bit or a qubit, in the end, the information is stored physically in classical or quantum computer hardware. So, one can say information is physical and erasure of information demands a physical erasure, in turn demanding energy. This, by nature, is a dissipative process, thus irreversible. This concept was formalised by Rolf Landauer in 1961.⁴

Any logical function on a classical computer can be implemented using only NAND gates, thus calling it a *universal gate*. These computations are typically irreversible. Considering just the NAND gate

$$(a, b) \rightarrow \neg(a \wedge b)$$

has two input bits and one output bit, and we can't recover a unique input from the output bit. As formally shown by Landauer, as we keep losing information, we need energy to operate the NAND gate. So, if you are given a limited number of batteries, then there is a theoretical limit to how long your computation can run.

Is there a way out of this energy crisis? What if all steps can be made reversible? Is it possible to do so? Charles Bennett precisely showed this in 1973, concluding that any computation can be performed using only reversible steps. Thus, no dissipation and no power expenditure, at least in principle. The (Toffoli) gate

$$(a, b, c) \rightarrow (a, b, c \oplus a \wedge b)$$

⁴You can read more about this and a related concept, "Maxwell's demon", from [Preskill \[1998\]](#)

is a reversible 3-bit gate. It flips the third bit if the first two bits have value 1 and does nothing otherwise. Notice the third output bit is nothing but the NAND of a and b if $c = 1$. This shows we can transform an irreversible computation to a reversible one by replacing the NAND gates with Toffoli gates. This, in principle, causes negligible energy dissipation.

Though it seems as if we can save a lot of energy by replacing NAND gates with Toffoli gates, note that in this process, we end up having 3 bits in the place of 2 bits previously used. If one wishes to erase all these extra junk bits, one needs to pay this energy cost later! But can we do something smart? Here comes the real power of reversible computing. When this occurred to Bennett, he pointed out that one can reverse the entire computation after getting the output and go back to the initial configuration. This restores the state of the initial bits, thus removing junk without extra energy expense!

Today's classical computer hardware can handle this energy dissipation, and we still continue to use irreversible computation. But eventually, when computer chips and components shrink in size, to protect them from melting due to this heat dissipation, we would need the help of reversible computing.

4.4.2 Classical Reversible Computation v.s Quantum Computing

Classically NOT is the only nontrivial reversible operation that can be performed on a single bit. Far more operations are possible on a single qubit. All linear combinations of reversible operations that take a single qubit to another single qubit state represent transformations \mathbf{U} , which are unitary and satisfy the condition

$$\mathbf{U}\mathbf{U}^\dagger = \mathbf{U}^\dagger\mathbf{U} = \mathbf{1}$$

By definition, unitary transformations are invertible. Thus, quantum computing is reversible.

On a classical computer, the permutation of the classical bits is a reversible operation. In fact, just applying the same permutation again gets back the classical bits to their initial position. The 2^n many n -bit configuration of a classical computer forms the basis of an n -qubit quantum state. Thus, any permutation \mathbf{P} of the 2^n bit states has an associated unitary transformation \mathbf{U} on n qubits. Therefore, we can just define \mathbf{U} to act like \mathbf{P} on the classical basis state, and this by linearity extends to an n -qubit state.

$$\mathbf{U}\mathbf{U}^\dagger = \mathbf{U}^\dagger\mathbf{U} = \mathbf{1}$$

Thus, on a classical computer, reversibility is an additional constraint, unlike the inherent reversibility of a quantum computer. Also, classical reversibility is restricted to the permutation of the bit strings, whereas in a quantum computer, it naturally allows for fundamentally quantum behaviour like superposition, entanglement, and non-classical correlations.

4.5 Quantum Parallelism

In chapter 3, we saw the powers Turing Machines have - anything that can be computed can be captured by a Turing machine. *Turing completeness* is the ability for a computational model or a system of instructions to simulate a Turing machine. A programming language that is Turing Complete is theoretically capable of expressing all tasks accomplished by computers; nearly all classical programming languages are Turing Complete if the limitations of finite memory are ignored. So, given that everything of interest can be computed on a classical computer, why are we so keen on building a quantum computer? Even energy dissipation can be avoided by classical reversible computation.

To understand what quantum computers can offer more than a classical computer, one has to understand what are the unique features of quantum mechanics that are not described in classical mechanics. Many would answer this with two keywords, *superposition* and *entanglement*.

Consider a two-qubit state $|0\rangle|0\rangle$. If we apply to each qubit the 1-qubit Hadamard transformation \mathbf{H} , then we get

$$\begin{aligned} (\mathbf{H} \otimes \mathbf{H})(|0\rangle \otimes |0\rangle) &= \mathbf{H}_1 \mathbf{H}_0 |0\rangle |0\rangle = (\mathbf{H}|0\rangle)(\mathbf{H}|0\rangle) \\ &= \frac{1}{\sqrt{2}}(|0\rangle + |1\rangle) \frac{1}{\sqrt{2}}(|0\rangle + |1\rangle) \\ &= \frac{1}{2}(|0\rangle|0\rangle + |0\rangle|1\rangle + |1\rangle|0\rangle + |1\rangle|1\rangle) \\ &= \frac{1}{2}(|0\rangle_2 + |1\rangle_2 + |2\rangle_2 + |3\rangle_2). \end{aligned}$$

The n -qubit state generalisation follows as,

$$\mathbf{H}^{\otimes n} |0\rangle_n = \frac{1}{2^{n/2}} \sum_{0 \leq x < 2^n} |x\rangle_n,$$

where

$$\mathbf{H}^{\otimes n} = \mathbf{H} \otimes \mathbf{H} \otimes \cdots \otimes \mathbf{H}, \quad n \text{ times.}$$

Suppose we are interested in constructing a function $f : \{0, 1\}^n \rightarrow \{0, 1\}^m$. If the initial state of the input register is $|0\rangle_n$, applying \mathbf{U}_f to this state directly computes $f(0^n)$. Instead, we apply an n -fold Hadamard transformation to that register first to get an equally weighted superposition of all possible n -qubit inputs. If we then apply \mathbf{U}_f to that superposition, by linearity, we get

$$\begin{aligned} \mathbf{U}_f(\mathbf{H}^{\otimes n} \otimes \mathbf{1}_m)(|0\rangle_n |0\rangle_m) &= \frac{1}{2^{n/2}} \sum_{0 \leq x < 2^n} \mathbf{U}_f(|x\rangle_n |0\rangle_m) \\ &= \frac{1}{2^{n/2}} \sum_{0 \leq x < 2^n} |x\rangle_n |f(x)\rangle_m. \end{aligned}$$

So if we have a thousand qubits in the input register, initially all in the state $|0\rangle_{1000}$ (and m more in the output register), applying thousand Hadamard then \mathbf{U}_f results in $2^{1000} \approx 10^{301}$ evaluations of the function f . Which is much larger than the number of atoms (10^{80}) in the universe! Thus, starting with just a thousand qubits, it seems as though we can do 10^{301} computations! This is called *quantum parallelism*.

So, can we now claim we have successfully computed $f(x)$ for all x in one run of our quantum circuit? The answer is no! And this clearly shows why quantum computing is not parallel computing.

Though $\frac{1}{2^{n/2}} \sum_{0 \leq x < 2^n} |x\rangle_n |f(x)\rangle_m$ holds the superposition of all values of $f(x)$ after measurement the state of the register reduces to $|x_0\rangle |f(x_0)\rangle$ and we no longer can know about $f(x)$ for any other x other than x_0 .

What if we now make a sufficiently large number of copies of the output register and measure all to get $f(x)$ at all x without running the whole computation over again? Unfortunately, this is also not possible, and we will see why in the next section.

4.6 No-Cloning Theorem

Theorem 4.6.1. (*No-Cloning Theorem*) *There exist no unitary \mathbf{U} such that $\mathbf{U}(|\psi\rangle|0\rangle) = |\psi\rangle|\psi\rangle$*

Proof. If

$$\mathbf{U}(|\psi\rangle|0\rangle) = |\psi\rangle|\psi\rangle \quad \text{and} \quad \mathbf{U}(|\phi\rangle|0\rangle) = |\phi\rangle|\phi\rangle$$

then it follows from linearity that

$$\mathbf{U}(a|\psi\rangle + b|\phi\rangle)|0\rangle = a\mathbf{U}|\psi\rangle|0\rangle + b\mathbf{U}|\phi\rangle|0\rangle = a|\psi\rangle|\psi\rangle + b|\phi\rangle|\phi\rangle$$

But if \mathbf{U} cloned arbitrary inputs, we would have

$$\begin{aligned} \mathbf{U}(a|\psi\rangle + b|\phi\rangle)|0\rangle &= (a|\psi\rangle + b|\phi\rangle)(a|\psi\rangle + b|\phi\rangle) \\ &= a^2|\psi\rangle|\psi\rangle + b^2|\phi\rangle|\phi\rangle + ab|\psi\rangle|\phi\rangle + ab|\phi\rangle|\psi\rangle \end{aligned}$$

Notice that these cross terms are missing in $\mathbf{U}|\phi\rangle|0\rangle = a|\psi\rangle|\psi\rangle + b|\phi\rangle|\phi\rangle$. These two equations are equal only when either a or b is zero. Thus, for an arbitrary state, we do not have any unitary that can copy that state. ■

Not only that, we can prove a stronger claim that we can not even copy an arbitrary state to a reasonable degree of approximation.

Theorem 4.6.2. *There exists no unitary \mathbf{U} that can approximately clone an arbitrary state. That is there is no unitary \mathbf{U} such that $\mathbf{U}(|\psi\rangle|0\rangle) \approx |\psi\rangle|\psi\rangle$*

Proof. Suppose that \mathbf{U} approximately cloned both $|\phi\rangle$ and $|\psi\rangle$:

$$\mathbf{U}(|\psi\rangle|0\rangle) \approx |\psi\rangle|\psi\rangle \quad \text{and} \quad \mathbf{U}(|\phi\rangle|0\rangle) \approx |\phi\rangle|\phi\rangle$$

Then, since unitary transformations preserve inner products, since the inner product of a tensor product of states is the (ordinary) product of their inner products, and since $\langle 0 | 0 \rangle = 1$, it follows that

$$\langle \phi | \psi \rangle \approx \langle \phi | \psi \rangle^2$$

But this requires $\langle \phi | \psi \rangle$ to be either close to 1 or close to 0. Hence a unitary transformation can come close to cloning both of two states $|\psi\rangle$ and $|\phi\rangle$ only if the states are very nearly the same, or very close to being orthogonal. In all other cases at least one of the two states will be badly copied. ■

If this were the whole picture, then we would not be having researchers and companies interested in quantum computing. Though we can not get all the values of $f(x)$ we can do something clever and interesting. Here, the skill and art of algorithmic thinking come into play. Alongside U_f , one could add several other unitaries cleverly such that when the final measurement is done, one could extract useful information about "relations" between the values of f for several different x , which a classical computer could get only by making several independent evaluations. The price one inevitably pays for this relational information is the loss of the possibility of learning the actual value $f(x)$ for any individual x . This tradeoff of one kind of information for another is typical of quantum computation and typical of quantum physics in general, where it is called the uncertainty principle, stated by Werner Heisenberg in the context of the position of a particle versus its momentum.

In the following chapters, we will have a glimpse into the art of designing algorithms in quantum computers with an advantage over their classical counterparts.

4.7 Building a Qubit

Among the diverse hardware platforms being explored for quantum computing, superconducting circuit architectures have emerged as a front-runner. These systems, formally known as *circuit quantum electrodynamics* (cQED)⁵ devices, harness the quantum dynamics of microwave photons and electrical currents within superconducting circuits to create, control, and read out quantum information. Their key advantage lies in their design flexibility and potential for scaling to large numbers of qubits.

Circuit QED devices are most often formed by embedding a special kind of superconducting device, known as a Josephson junction, into complex systems of circuitry, further embedded with superconductors for minimising dissipative losses. Josephson junction⁶ can be intuitively imagined as a non-linear induction circuit, which helps in the physical realisation of quantum states. It is made by sandwiching a thin layer of a nonsuperconducting material between two layers of superconducting material.

⁵For an excellent in-depth review, refer to the paper [Roth et al. \[2023\]](#) or the lecture notes [Girvin et al. \[2014b\]](#).

⁶The 2025 Nobel Prize in physics has been awarded to John Clarke, Michel H. Devoret and John M. Martinis *for the discovery of macroscopic quantum mechanical tunnelling and energy quantisation in an electric circuit.*, check out [Physics Nobel – 2025 : Schrödinger’s cat and her laboratory cousins](#)

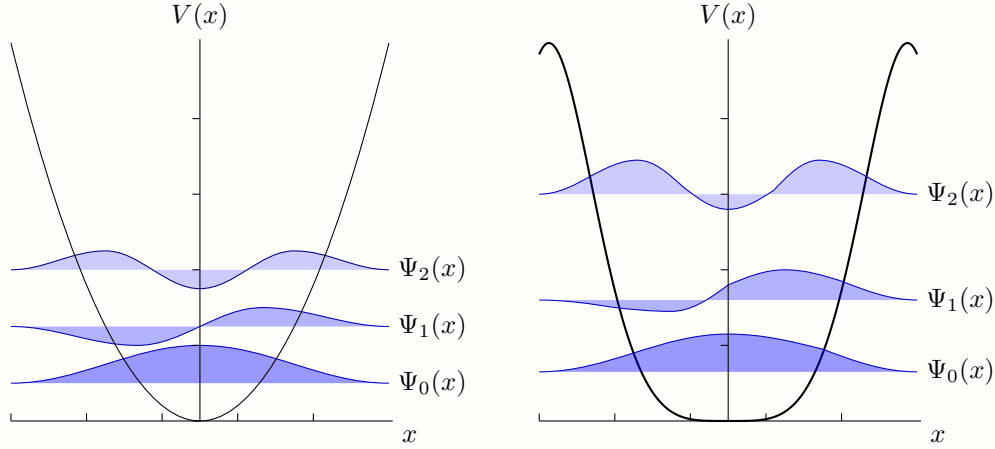


Figure 4.3: LC oscillators: Harmonic and Anharmonic oscillators.

To appreciate the role of the Josephson junction, one must first grasp the fundamentals of superconductivity. When certain metals and alloys are cooled to extremely low temperatures (typically within a few degrees of absolute zero), they undergo a phase transition. At a specific critical temperature, the material shifts from its normal, resistive state to a superconducting state, where direct electrical current flows with zero resistance. Below this temperature, the subtle interaction between electrons and the crystal lattice of the metal becomes attractive, allowing electrons to overcome their natural repulsion and bind together into what are known as Cooper pairs.

The formation of Cooper pairs opens an energy gap, creating a collective, macroscopic quantum state, a superfluid of charge that moves without resistance. It is by quantising the collective electrical degrees of freedom of this superfluid, such as the number of Cooper pairs on an isolated superconducting island or the magnetic flux threading a loop, that we can engineer a robust two-level quantum system: the physical realisation of a qubit.

Numerous types of superconducting qubits exist, including the charge qubit, flux qubit, phase qubit, and fluxonium, each differing in its design and energy scales. A particularly interesting variant is the transmon, a type of charge qubit formed by two superconducting islands connected by a Josephson junction. In its simplest form, the transmon is an LC oscillator, a parallel combination of a capacitor and an inductor. This additionally gets rid of external fields in the circuitry through the superconducting nature and reduces it to a single oscillatory degree of freedom.

However, a simple LC circuit is a quantum harmonic oscillator, characterized by an infinite ladder of equally spaced energy levels. This is unsuitable for a qubit, as a control signal intended for one transition would excite all of them. This is where the Josephson junction's nonlinearity becomes critical. By replacing the standard linear inductor with a junction, the circuit's potential energy is no longer a simple quadratic function but instead follows

a cosine dependence on the magnetic flux. This property, known as anharmonicity, breaks the uniform spacing of the energy levels. It ensures that the energy gap between the ground state ($|0\rangle$) and the first excited state ($|1\rangle$) is unique, allowing us to selectively address the $|0\rangle \leftrightarrow |1\rangle$ transition with precisely tuned microwave pulses, thereby realizing a high-fidelity qubit. Fig. 4.3 displays a LC oscillator circuit solution behaving as a harmonic oscillator with equally spaced energy levels, and an equivalent LC oscillator solution with an inductor replaced by a Josephson junction behaving as an anharmonic oscillator.

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Part II

Quantum Computing

Chapter 5

Basic Quantum Algorithms

“For me, great algorithms are the poetry of computation. Just like verse, they can be terse, allusive, dense, and even mysterious. But once unlocked, they cast a brilliant new light on some aspect of computing.”

– Francis Sullivan, *The Joy of Algorithms*

5.1 Some Basic Functions

As we saw in chapter 4, the way to construct a function f is by constructing a unitary \mathbf{U}_f such as the one shown in Fig. 5.1. If $y = 0$ then the output register holds $|x\rangle \otimes |f(x)\rangle$, thus we have computed $f(x)$. In this section, we will see how to design \mathbf{U}_f , by using the basic gates we defined in Sec. 4.3 for some simple functions f . Also, recall that the function is from $\{0, 1\}^n$ to $\{0, 1\}$.

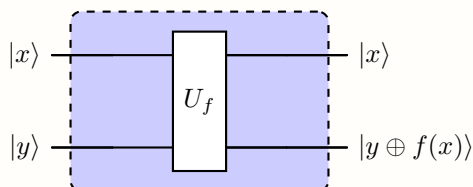


Figure 5.1: Unitary to represent a function

5.1.1 Constant Function

Let's first look at the following table:

In order to construct any function, it helps to write such a table and then think what should \mathbf{U}_f be so that the output register is of the form $|x\rangle \otimes |y \oplus f(x)\rangle$.

x	$f(x)$	$0 \oplus f(x)$	$1 \oplus f(x)$
0	0	0	1
1	0	0	1

$$f(x) = 0$$

Now, it is easy to see why $\mathbf{U}_f = \mathbb{I}$ works in this case.



Figure 5.2: Circuit representing constant function $f(x) = 0$

What if $f(x)$ is 1?

Doing the same exercise done for $f(x) = 0$ again we find:

x	$f(x)$	$0 \oplus f(x)$	$1 \oplus f(x)$
0	1	1	0
1	1	1	0

$$f(x) = 1$$

Thus, the same circuit given for $f(x) = 0$ works for $f(x) = 1$.

5.1.2 Identity Function

Again, let's look at the table:

x	$f(x)$	$0 \oplus f(x)$	$1 \oplus f(x)$
0	0	0	1
1	1	1	0

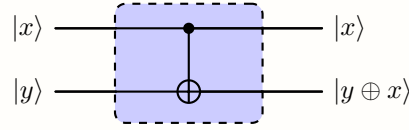
$$f(x) = x$$

Now, it is easy to see why $\mathbf{U}_f = \text{cNOT}$ works in this case.

Is cloning achieved with the Identity function?

The answer is No! If f is the *identity function*, $f(x) = x$. Thereby, the unitary is

$$\mathbf{U}_f(|x\rangle|0\rangle) = |x\rangle|x\rangle$$

Figure 5.3: Circuit representing identity function $f(x) = x$

Now, let $|x\rangle = a|0\rangle + b|1\rangle$. We have the action of the unitary,

$$\begin{aligned} \mathbf{U}_f|x\rangle|0\rangle &= \mathbf{U}_f((a|0\rangle + b|1\rangle)(|0\rangle)) = \mathbf{U}_f(a|00\rangle + b|10\rangle) \\ &= a|0\rangle|0 \oplus f(0)\rangle + b|1\rangle|0 \oplus f(0)\rangle \\ &= a|0\rangle|0\rangle + b|1\rangle|1\rangle = (a|00\rangle + b|11\rangle) \end{aligned}$$

However, for the product state,

$$(a|0\rangle + b|1\rangle)(a|0\rangle + b|1\rangle) = (a^2|00\rangle + ab|01\rangle + ba|10\rangle + b^2|11\rangle)$$

Thereby, we do not have the cross terms from the unitary. Therefore, we have not cloned $|x\rangle$.

5.1.3 Swap

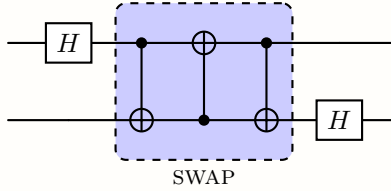


Figure 5.4: Circuit representing Swap operation

5.1.4 Is such a unitary always possible?

The computational process generally requires more registers apart from the input and output registers for workspace. In general, the input and output registers will become entangled with the states of the additional r qubits. In this case, we cannot have a unitary \mathbf{U}_f that relates only the input and output state, as shown in 5.1. If the action of the computer on all $n + m + r$ qubits has a special form such that at the end of the computation, the workspace registers are not entangled with the input-output qubits and have a state that is independent of the initial state of the input and output qubit, then having such a unitary \mathbf{U}_f is possible.

A quantum unitary \mathbf{W}_f applies to *all* the registers, say $n + m + r$, the input, output, and workspace. One can achieve this by simply taking advantage of the fact that unitary transformations are reversible.

- Apply a unitary \mathbf{V}_f only on $n + r$ qubits and store $f(x)$ in m -qubits of the $n + m$ qubits. As the m output qubits are untouched, they are not entangled with the input and workspace qubits.
- Change output register y to $y \oplus f(x)$ by applying m cNOT gates, \mathbf{C}_m .
- Since the state of the $n + r$ qubits is not altered by the application of \mathbf{C}_m , we can inverse the transformation \mathbf{V}_f^\dagger to restore them to their original state.

Note that in the above process (also shown in Fig. 5.5), as the workspace registers are restored back, they are neither entangled nor dependent on the input and output states. Thus, we can safely use the above trick and talk about \mathbf{U}_f (as in Fig. 5.1) every time.

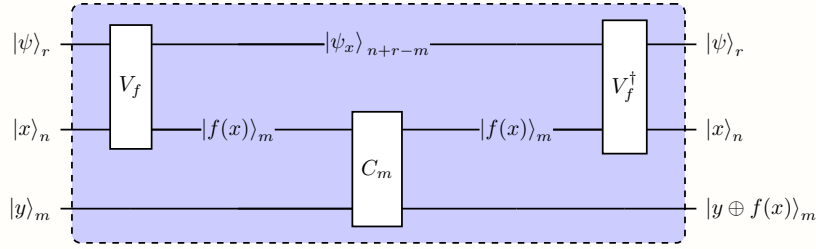


Figure 5.5: Circuit to disentangle workspace registers

To illustrate this, let us examine the example of the quantum 1-bit adder circuit. The circuit, along with a carry qubit, which is the additional work space qubit, is given in figure 5.6. One can verify that this circuit performs the bitwise addition of the two qubits $|x_1\rangle$ and $|x_2\rangle$, storing the carry bit as $|C\rangle$, and final sum as $|S\rangle$, replacing $|x_2\rangle$.

To write this circuit with a disentangled workspace qubit, we add cNOT with V_f and V_f^\dagger parts, as shown in figure 5.7.

5.2 Deutsch's Problem

PROBLEM STATEMENT: Given a function $f : \{0, 1\} \rightarrow \{0, 1\}$ find if f is a constant function or not.

CLASSICAL ALGORITHM: Compute the value of f on both 0 and 1 and compare the values. This takes two computations of the function f .

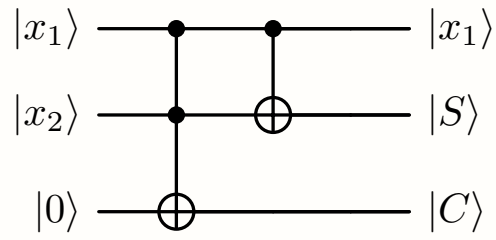


Figure 5.6: Quantum 1-bit Half Adder

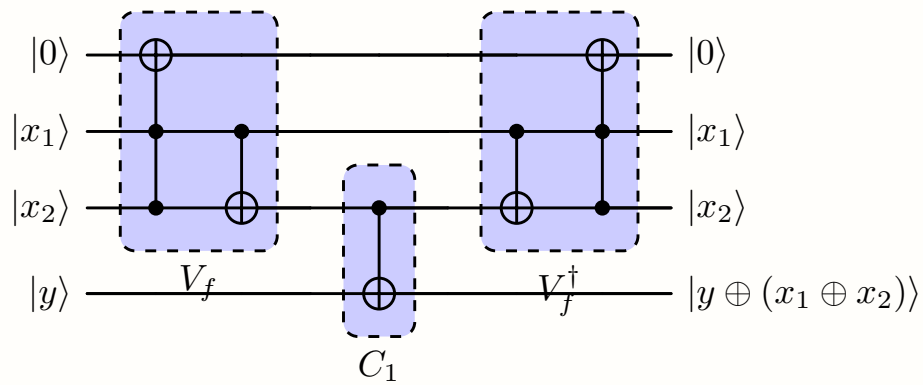


Figure 5.7: 1-Bit adder with disentangled workspace register

One can not do any better on a classical computer. But with a quantum computer, can we do better? That is, find if $f(0)$ and $f(1)$ are different just with one query to U_f ?

QUANTUM ALGORITHM:

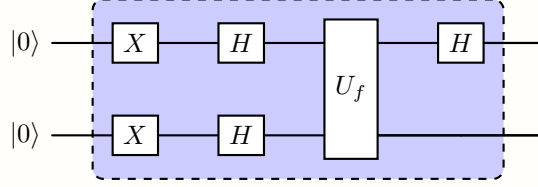


Figure 5.8: Deutsch's Algorithm

Therefore, the overall action of these gates is,

$$\begin{aligned} & (\mathbf{H} \otimes \mathbf{1}) \mathbf{U}_f (\mathbf{H} \otimes \mathbf{H}) (\mathbf{X} \otimes \mathbf{X}) (|0\rangle|0\rangle) \\ &= \begin{cases} |1\rangle \frac{1}{\sqrt{2}} (|f(0)\rangle - |\tilde{f}(0)\rangle), & f(0) = f(1), \\ |0\rangle \frac{1}{\sqrt{2}} (|f(0)\rangle - |\tilde{f}(0)\rangle), & f(0) \neq f(1). \end{cases} \end{aligned}$$

Thus, the state of the input register determines whether the function is constant or not, with a single computation of the functional value of the input state $|00\rangle$.

DISCUSSION: Note that there are only 4 possible functions $f : \{0,1\} \rightarrow \{0,1\}$ represented by the 4 circuits given below 5.9. Suppose we are given this U_f as a black box; that is, we do not know which one of these four is our function; how will we find f ? We can let the black box act twice, once on the state $|0\rangle|0\rangle$ and once on $|1\rangle|0\rangle$ and find f . Similar to when we have a classical black box that computes the value of f , we need to query it twice to find $f(0)$ and $f(1)$ in order to say if the function is constant or not.

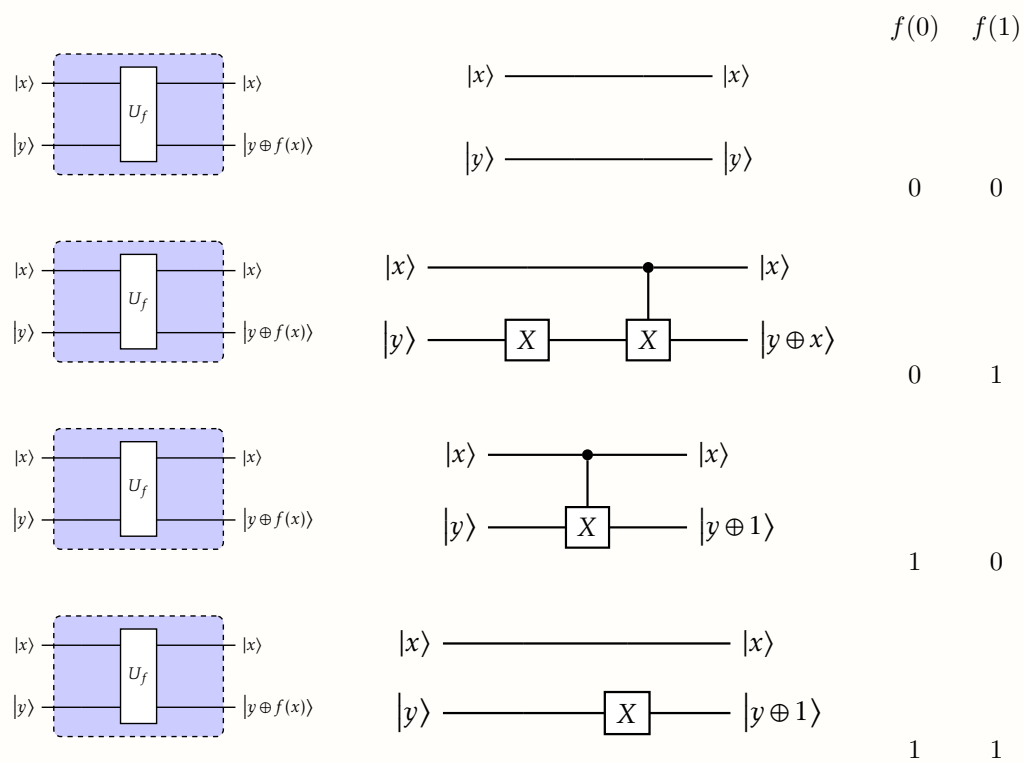
Instead, by creating an equal superposition of $|0\rangle$ and $|1\rangle$, we can create an equal superposition of $f(0)$ and $f(1)$ with one call to the oracle. But note that if you measure the state now, it collapses to either $f(0)$ or $f(1)$, and from the 4 circuits in Fig. 5.9, we can only narrow down to 2, but still, we have equal probability of the function being constant or not.

We do not wish to know the exact functional value of $f(0)$ or $f(1)$. So cleverly we trade off this information to get to know whether f is constant or balanced. Consider applying the following gates to the input qubits,

$$\begin{aligned} (\mathbf{H} \otimes \mathbf{H}) (\mathbf{X} \otimes \mathbf{X}) (|0\rangle|0\rangle) &= (\mathbf{H} \otimes \mathbf{H}) (|1\rangle|1\rangle) = \left(\frac{1}{\sqrt{2}}|0\rangle - \frac{1}{\sqrt{2}}|1\rangle \right) \left(\frac{1}{\sqrt{2}}|0\rangle - \frac{1}{\sqrt{2}}|1\rangle \right) \\ &= \frac{1}{2} (|0\rangle|0\rangle - |1\rangle|0\rangle - |0\rangle|1\rangle + |1\rangle|1\rangle). \end{aligned}$$

And now apply U_f

$$\frac{1}{2} \left(U_f(|0\rangle|0\rangle) - U_f(|1\rangle|0\rangle) - U_f(|0\rangle|1\rangle) + U_f(|1\rangle|1\rangle) \right).$$

Figure 5.9: Circuits showing all possible functions $f : \{0, 1\} \rightarrow \{0, 1\}$.

$$\frac{1}{2}(|0\rangle|f(0)\rangle - |1\rangle|f(1)\rangle - |0\rangle|\tilde{f}(0)\rangle + |1\rangle|\tilde{f}(1)\rangle),$$

where, $\tilde{x} = 1 \oplus x$ so that $\tilde{1} = 0$ and $\tilde{0} = 1$, and $\tilde{f}(x) = 1 \oplus f(x)$. So if $f(0) = f(1)$ the output state is

$$\frac{1}{2}(|0\rangle - |1\rangle)(|f(0)\rangle - |\tilde{f}(0)\rangle), \quad f(0) = f(1),$$

but if $f(0) \neq f(1)$ then $f(1) = \tilde{f}(0)$, $\tilde{f}(1) = f(0)$, and the output state becomes

$$\frac{1}{2}(|0\rangle + |1\rangle)(|f(0)\rangle - |\tilde{f}(0)\rangle), \quad f(0) \neq f(1).$$

Note that the first qubit in both these cases is orthogonal to the other. Also, they are nothing but the eigenvalues of the Pauli X operator. At this state, one could either directly measure the 1st qubit in the X basis or apply the Hadamard transformation to the input register and measure in the standard computational basis.

$$|1\rangle \frac{1}{\sqrt{2}}(|f(0)\rangle - |\tilde{f}(0)\rangle), \quad f(0) = f(1),$$

$$|0\rangle \frac{1}{\sqrt{2}}(|f(0)\rangle - |\tilde{f}(0)\rangle), \quad f(0) \neq f(1).$$

Remarks. One could just apply the X gate to the second qubit and not the first qubit. In this case, the same calculation holds, and the only change is that the value of the first qubit will be $|0\rangle$ when f is constant and $|1\rangle$ when f is balanced.

Remarkably, with a quantum computer, we did not have to run \mathbf{U}_f twice to determine whether or not f is constant. We could do this in a single run. Interestingly, when we did this, we learned nothing about the individual values of $f(0)$ and $f(1)$, but we were nevertheless able to answer the question about their relative values: whether or not they are the same. Thus, we get less information than we get in answering the question with a classical computer, but by renouncing the possibility of acquiring that part of the information that is irrelevant to the question we wish to answer, we can get the answer with only a single application of the black box.

In Fig. 5.9, we saw the equivalent circuit representation of the 4 possible functions $f : \{0, 1\} \rightarrow \{0, 1\}$. Applying Hadamard gates to each qubit, both before and after the application of \mathbf{U}_f , must produce exactly the same result as it would if the Hadamards were applied to the equivalent circuits. After applying Hadamard, the resulting circuit will look like the circuit shown in Fig. 5.10

The Fig. 5.10 shows explicitly that when \mathbf{U}_f is sandwiched between Hadamards, the input register ends up in the state $|0\rangle$ if $f(0) = f(1)$ and in state $|1\rangle$ if $f(0) \neq f(1)$

Hadamard swaps the control and target

Notice that in the Circuit 5.10, after application of the Hadamard operator, the control and target qubits of the cNOT gate have swapped.

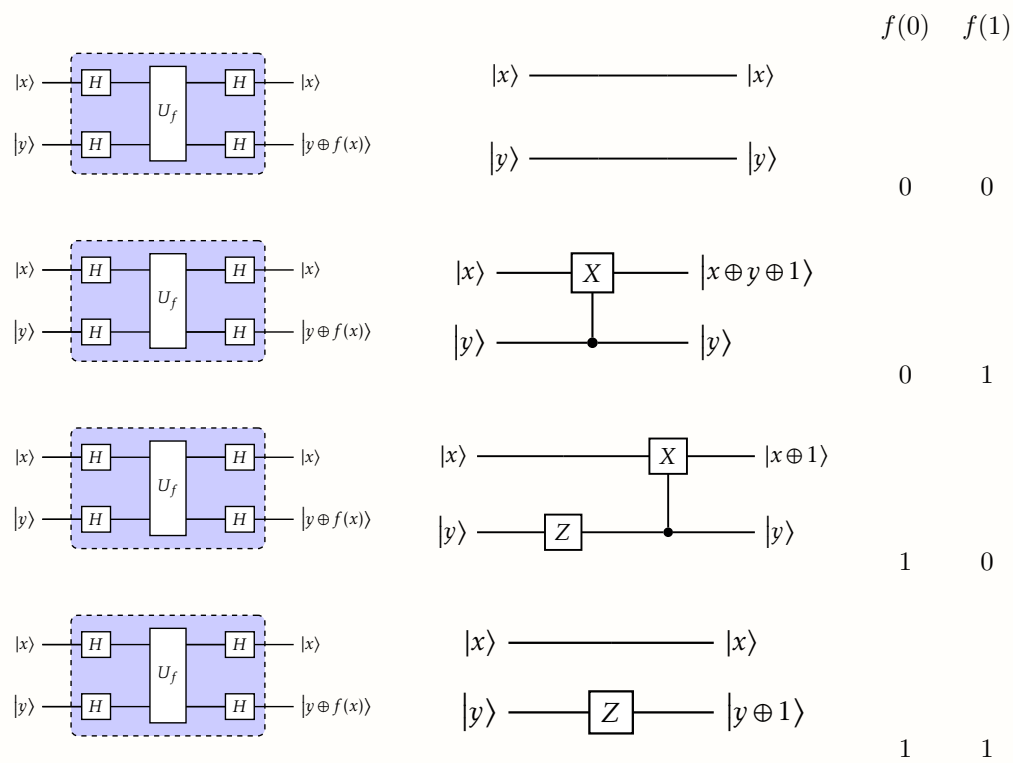


Figure 5.10: Equivalent circuit for Deutsch algorithm

5.3 Deutsch–Jozsa Problem

Deutsch–Jozsa Problem is a natural extension of Deutsch’s Problem to an n -bit function. We say $f : \{0, 1\}^n \rightarrow \{0, 1\}$ is *constant* if $f(x) = c$ for all $x \in \{0, 1\}^n$, where c is a fixed constant which is either 0 or 1. And we say f is *balanced* when exactly half of the 2^n inputs map to 0 and the other half map to 1.

PROBLEM STATEMENT: Given a function a n -bit function, $f : \{0, 1\}^n \rightarrow \{0, 1\}$, that is promised to be either constant or balanced. Determine if f is constant or balanced?

CLASSICAL ALGORITHM: We should compute the function on at least $2^{n-1} + 1$ inputs, that is, at least 1 more than half the inputs, in the "worst-case" to find if f is constant or balanced. If we are lucky on the second or third computation of f , we might find that the functional value of f does not match the previously computed functional values of f . In this case, we can directly halt after just 2 to 3 computations of the function. But in the *worst-case* it may so happen that all the first 2^{n-1} inputs turn out to be the same value. In this case, unless we compute at least one more input, we cannot certainly say if the function is constant or balanced. Thus, a classical computer needs at least $2^{n-1} + 1$ queries.

Can we do better? Like Deutsch’s problem, is there a way to magically get to know this in just one query on a quantum computer? The answer turns out to be yes!

QUANTUM ALGORITHM:

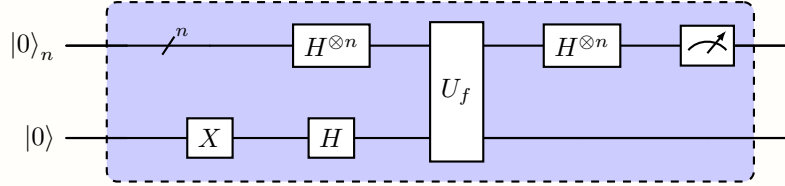


Figure 5.11: Deutsch–Jozsa Algorithm

Deutsch–Jozsa algorithm is just the application of the Deutsch algorithm, but on n input qubits as opposed to 1.

Hadamard on n -qubits

The action of \mathbf{H} on a single qubit can be compactly summarised as

$$\mathbf{H}|x\rangle_1 = \frac{1}{\sqrt{2}} (|0\rangle + (-1)^x |1\rangle) = \frac{1}{\sqrt{2}} \sum_{y=0}^1 (-1)^{xy} |y\rangle$$

If we apply $\mathbf{H}^{\otimes n}$ to an n -qubit computational-basis state $|x\rangle_n$ we can therefore

express the result as

$$\begin{aligned} \mathbf{H}^{\otimes n} |x\rangle_n &= \frac{1}{2^{n/2}} \sum_{y_{n-1}=0}^1 \cdots \sum_{y_0=0}^1 (-1)^{\sum_{j=0}^{n-1} x_j y_j} |y_{n-1}\rangle \cdots |y_0\rangle \\ &= \frac{1}{2^{n/2}} \sum_{y=0}^{2^n-1} (-1)^{x \cdot y} |y\rangle_n \end{aligned}$$

Doing the calculation similar to how we did for the Deutsch problem, the final state of the first n -qubits will be

$$\sum_z \sum_x \frac{(-1)^{x \cdot z + f(x)} |z\rangle}{2^n} \left[\frac{|0\rangle - |1\rangle}{\sqrt{2}} \right]$$

Note that if f is a constant function, then the amplitude of $|0\rangle^{\otimes n}$ is ± 1 depending on the value of f . This means that when measuring the first n -qubits, the probability of getting anything except all 0s is zero. But if the function is balanced, equal 1s and 0s will make the amplitude of $|0\rangle^{\otimes n}$ vanish. Thus, in this case, the probability of getting all 0s on measuring the first n -qubits is zero. Thus we are able to say if the function is balanced or constant in one computation of f .

5.4 Bernstein Vazirani Problem

PROBLEM STATEMENT: $f : \{0, 1\}^n \rightarrow \{0, 1\}^n$ such that $f(x) = x \cdot a$ ($x \cdot a$ is bitwise modulo-2 inner product). The goal is to find a , given we have a black box that evaluates $f(x) = x \cdot a$.

CLASSICAL ALGORITHM: The m^{th} bit of a is $a \cdot 2^m$, since the binary expansion of 2^m has 1 in position m and 0 in all the other positions. So with a classical computer, we can learn the n bits of a by applying f to the n values $x = 2^m, 0 \leq m < n$. As only one bit of information is provided in each query, no classical algorithm can do better than this.

Whereas with a quantum computer, a single invocation is enough to determine a completely, regardless of how big n is!

QUANTUM ALGORITHM:

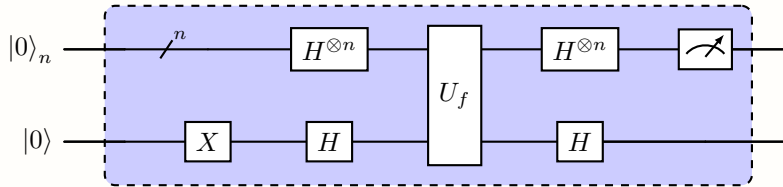


Figure 5.12: Bernstein Vazirani Algorithm

The overall circuit does the following,

$$\mathbf{H}^{\otimes(n+1)} \mathbf{U}_f \mathbf{H}^{\otimes(n+1)} |0\rangle_n |1\rangle_1 = |a\rangle_n |1\rangle_1,$$

where measuring the first n -qubits gives the required a in one computation of the function. **DISCUSSION:** This time, let us take a circuit theory approach to design the necessary quantum circuit. Suppose $a = 10010$, here our function f is assumed to be an $n = 4$ bit function. When $f(x) = a \cdot x$, the action of \mathbf{U}_f on the computational basis is to flip the 1-qubit output register once, whenever a bit of x and the corresponding bit of a are both 1. When the state of the input register is $|x\rangle_n$ this action can be performed by a collection of cNOT gates all targeted on the output register. There is one cNOT for each nonzero bit of a , controlled by the qubit representing the corresponding bit of x . The combined effect of these cNOT gates on every computational basis state is precisely that of \mathbf{U}_f . Therefore, the effect of any other transformations preceding and/or following \mathbf{U}_f can be understood by examining their effect on this equivalent collection of cNOT gates, even though \mathbf{U}_f may actually be implemented in a completely different way.

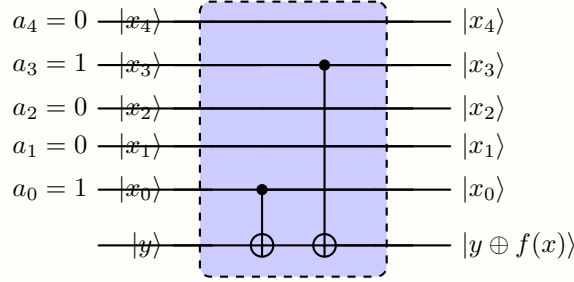


Figure 5.13: Circuit of $f(x) = x \cdot a$ for a given $a = 10010$

Since we have no control over what the value of a can be, we wish to take away the control from a (pun intended!). We have seen that we can flip the control and target qubits by application of Hadamard gates.

After this reversal of target and control qubits, the output register controls every one of the cNOT gates, and since the state of the output register is $|1\rangle$, every one of the NOT operators acts. That action flips just those qubits of the input register for which the corresponding bit of a is 1. Since the input register starts in the state $|0\rangle_n$, this changes the state of each qubit of the input register to $|1\rangle$, if and only if it corresponds to a nonzero bit of a . As a result, the state of the input register changes from $|0\rangle_n$ to $|a\rangle_n$.

Algebraically, as always, we start with an equal superposition of the input registers and apply \mathbf{U}_f . Now we apply Hadamard again as this will give the form $x \cdot (a + y)$ as an exponent to -1 , helping to capture only those y s that are equal to a .

$$(\mathbf{H}^{\otimes n} \otimes \mathbf{1}) \mathbf{U}_f (\mathbf{H}^{\otimes n} \otimes \mathbf{H}) |0\rangle_n |1\rangle_1$$

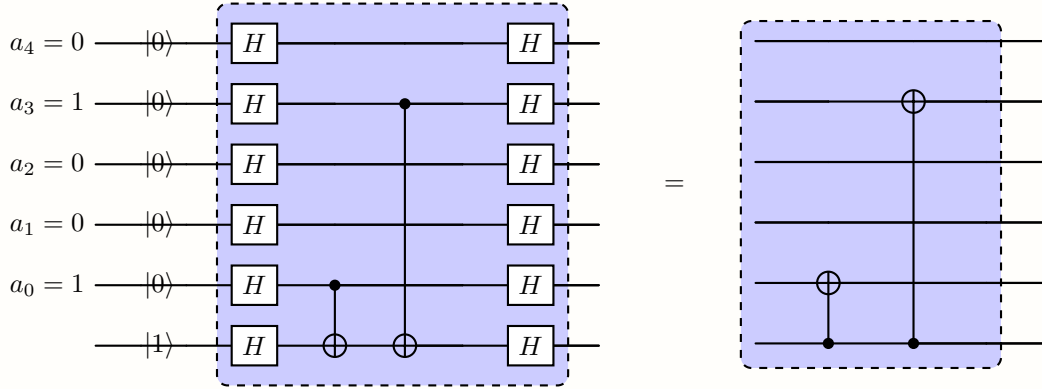


Figure 5.14: Sandwiching cNOT between Hadamard

$$\begin{aligned}
 &= (\mathbf{H}^{\otimes n} \otimes \mathbf{1}) \mathbf{U}_f \left(\frac{1}{2^{n/2}} \sum_{x=0}^{2^n-1} |x\rangle \right) \frac{1}{\sqrt{2}} (|0\rangle - |1\rangle) \\
 &= \frac{1}{2^{n/2}} \left(\mathbf{H}^{\otimes n} \sum_{x=0}^{2^n-1} (-1)^{f(x)} |x\rangle \right) \frac{1}{\sqrt{2}} (|0\rangle - |1\rangle) \\
 &= \frac{1}{2^n} \sum_{x=0}^{2^n-1} \sum_{y=0}^{2^n-1} (-1)^{f(x)+x \cdot y} |y\rangle \frac{1}{\sqrt{2}} (|0\rangle - |1\rangle).
 \end{aligned}$$

We do the sum over x first. If the function $f(x)$ is $a \cdot x$ then this sum produces the factor

$$\sum_{x=0}^{2^n-1} (-1)^{(a-x) \cdot y} = \prod_{j=1}^n \sum_{x_j=0}^1 (-1)^{(a_j+y_j)x_j}$$

At least one term in the product vanishes unless every bit y_j of y is equal to the corresponding bit a_j of a , i.e. unless $y = a$. Therefore, the entire computational process reduces to

$$\mathbf{H}^{\otimes(n+1)} \mathbf{U}_f \mathbf{H}^{\otimes(n+1)} |0\rangle_n |1\rangle_1 = |a\rangle_n |1\rangle_1,$$

Final \mathbf{H} to the 1-qubit output register to make the final expression look neater and more symmetric.

Thus, all n bits of the number a can now be determined by measuring the input register, even though we have called the subroutine only once!

Why can not we do the same swapping cNOT technique classically?

Interestingly, quantum computers can do this only because it allows the reversal of the control and target qubits of a cNOT operation solely by means of 1-qubit

(Hadamard) gates. One can also reverse control and target bits of a cNOT classically, but this requires the use of 2-qubit SWAP gates, rather than 1-qubit Hadamards. You can confirm for yourself that this circuit-theoretic solution to the Bernstein-Vazirani problem no longer works if one tries to replace all the Hadamard gates by any arrangement of SWAP gates.

5.5 Simon's Problem

PROBLEM STATEMENT: $f : \{0, 1\}^n \rightarrow \{0, 1\}^{n-1}$, a two to one function, such that $f(x \oplus a) = f(x)$. Find the period a , given a black box that computes f .

CLASSICAL ALGORITHM: With a classical computer, we can keep computing f until we by chance encounter x_i and x_j such that both give the same $f(x)$. Then we know that $a = x_i \oplus x_j$. At any stage of this process, if we pick m different x_k s such that none have the same functional value, all we can say is, for any pair $x_i \oplus x_j \neq a$. In this way we can reject at most $\binom{m}{2} = \frac{m(m-1)}{2}$ values of a . There are 2^{n-1} possible values for a , so m should be as big as 2^{n-1} to narrow down to one value of a . So, it needs exponentially many calls to the black box to compute a . Whereas we can compute a with just linear calls using a quantum computer.

QUANTUM ALGORITHM:

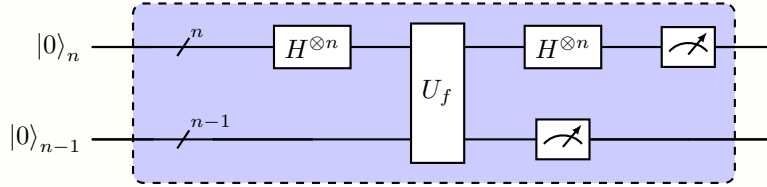


Figure 5.15: Simon's Algorithm

At every invocation of U_f , we get non-trivial information about a . With $\mathcal{O}(n)$ many invocations, we can with high probability find a .

DISCUSSION: Let's start with an equal superposition of all input states. On applying U_f to this, we get each of the terms in the equal superposition to be of the form $|x\rangle |f(x)\rangle$. Now, if we measure it collapses to a particular $f(x_0)$ and two different x , that is, x_0 and $x_0 \oplus a$, gives the same $f(x_0)$ (as f is a two-to-one function mapping n bit strings to $n-1$ bit strings). So, on measurement of the output qubit, the input qubit now collapses to an equal superposition of these two values of x .

$$\frac{1}{\sqrt{2}} (|x_0\rangle + |x_0 \oplus a\rangle)$$

With the input register in the above state, we apply the n -fold Hadamard transformation

$\mathbf{H}^{\otimes n}$.

$$\mathbf{H}^{\otimes n} \frac{1}{\sqrt{2}} (|x_0\rangle + |x_0 \oplus a\rangle) = \frac{1}{2^{(n+1)/2}} \sum_{y=0}^{2^n-1} \left((-1)^{x_0 \cdot y} + (-1)^{(x_0 \oplus a) \cdot y} \right) |y\rangle.$$

Since $(-1)^{(x_0 \oplus a) \cdot y} = (-1)^{x_0 \cdot y} (-1)^{a \cdot y}$, the coefficient of $|y\rangle$ is 0 if $a \cdot y = 1$ and $2(-1)^{x_0 \cdot y}$ if $a \cdot y = 0$. Therefore

$$\frac{1}{2^{(n-1)/2}} \sum_{a \cdot y = 0} (-1)^{x_0 \cdot y} |y\rangle$$

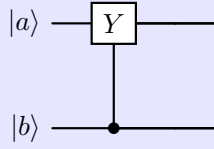
where the sum is now restricted to those y for which the modulo-2 bitwise inner product $a \cdot y$ is 0 rather than 1. So, if we now measure the input register, we learn (with equal probability) any of the values of y for which $a \cdot y = 0$ i.e. for which

$$\sum_{i=0}^{n-1} y_i a_i = 0 \pmod{2}$$

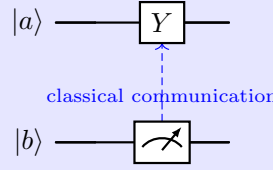
where a_i and y_i are corresponding bits in the binary expansions of a and y . Thus, each time we get non-trivial information about a . One can prove that with $\mathcal{O}(n)$ many invocations of \mathbf{U}_f with high probability one can find the value of a .¹

To avoid non-local application of unitary

Most often, it is hard to have control gates between qubits that are spatially separated. In this case, we can measure the control qubit and classically communicate its value to the target qubit. This can be done because figures (i) and (ii) are essentially equivalent.



(i) non-local controlled unitary



(ii) classical communication

To see their equivalence, consider a general 2-qubit state:

$$\begin{aligned} & a_{00}|00\rangle + a_{10}|10\rangle + a_{01}|01\rangle + a_{11}|11\rangle \\ &= (a_{00}|0\rangle + a_{10}|1\rangle) |0\rangle + (a_{01}|0\rangle + a_{11}|1\rangle) |1\rangle \\ &\equiv A_0 \frac{(a_{00}|0\rangle + a_{10}|1\rangle)}{A_0} |0\rangle + A_1 \left(\frac{a_{01}|0\rangle + a_{11}|1\rangle}{A_1} \right) |1\rangle \\ &|\psi\rangle \equiv A_0 |\phi_0\rangle |0\rangle + A_1 |\phi_1\rangle |1\rangle \end{aligned}$$

Where A_0 and A_1 are appropriate normalization constants.

¹For more detail on the complexity refer to the paper [Koiran et al. \[2007\]](#)

Non-local application of controlled unitary

$$U|\psi\rangle = A_0 |\phi_0\rangle |0\rangle + A_1 V_1 |\phi_1\rangle |1\rangle$$

where $U = \mathbb{I} \otimes V_1$. If $b = 1$, then the 1st qubit collapses to $V_1 |\phi_1\rangle$ and this happens with probability $|A_1|^2$. If $b = 0$, then the 1st qubit collapses to $|\phi_0\rangle$ with probability $|A_0|^2$.

Measurement and classical communication

If we measure $|b\rangle$ then we will get 1 with probability $|A_1|^2$ and 0 with probability $|A_0|^2$. When we communicate this information classically to Y . This also gives the same probabilities for $V_1 |\phi_1\rangle$ and $|\phi_0\rangle$, showing the equivalence.

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Chapter 6

Quantum Fourier Transform and Shor's Algorithm

"There cannot be a language more universal and more simple, more free from errors and obscurities, more worthy to express the invariable relations of all natural things[than mathematics]. [It interprets] all phenomena by the same language, as if to attest the unity and simplicity of the plan of the universe, and to make still more evident that unchangeable order which presides over all natural causes."

– Joseph Fourier, *The Analytical Theory of Heat*

The problem of how to factor a large integer efficiently has been studied extensively in number theory. It is generally believed that factorization of a number n is hard to do in an efficient way. That is, it cannot be done in a number of steps which is polynomial in the length of the integer we're trying to factor. The RSA cryptosystem, among others, relies on the presumed difficulty of this task. Classically, the fastest known algorithm is the Number Field Sieve algorithm, which works in super-polynomial but sub-exponential time, $\mathcal{O}(e^{n^{1/3}(\log n)^{2/3}})$.

In 1994, Peter Shor discovered an algorithm that can factor numbers in polynomial time, $\mathcal{O}(n^2 \log n \log \log n)$, using a quantum computer, a drastic improvement over the existing classical algorithms. That is, a quantum computer can factor a number exponentially faster than the best-known classical algorithms.

6.1 RSA Cryptography

For HTTPS on your browser, password managers, VPNs, financial banking, credit card chip and software licensing, RSA is one of the currently used and most effective public key cryptography protocols. Can a quantum computer break this protocol, which is robust to classical computers? Fortunately or unfortunately the answer turns out to be yes!

Suppose Alice and Bob want to communicate. Both Alice and Bob have two keys each, one

a public key that is publicly available to everyone and a private key that no one other than the owner knows. One possible scheme to communicate securely is as follows:

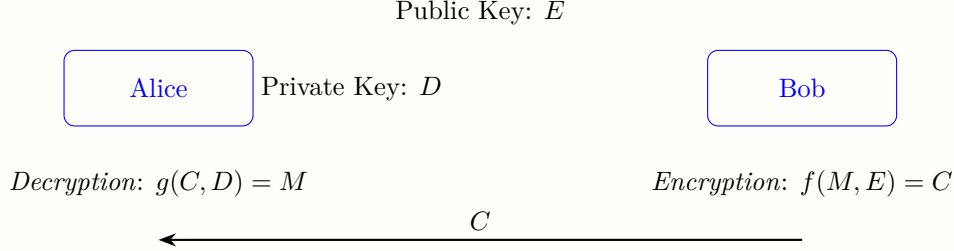


Figure 6.1: Public Key Cryptography

Bob encrypts his message through the function f , invoking the public key E , and sends the encrypted message C to Alice. Alice uses the function g to decrypt the message with the help of her private key D to recover the message. The functions f and g are released publicly as a part of the protocol. Public key cryptography works on the basis that the function f is extremely difficult to invert; that is, getting the message M from the cipher text C is extremely hard. But this becomes easy with D , the private key. Thus, such protocols heavily rely on the computational hardness of a problem.

The keys for the RSA algorithm are generated in the following way:

- Choose P and Q very large primes. Compute $N = PQ$.
- N is released as a part of the public key. N will be used as the modulo arithmetic for both public and private keys. Its length, usually expressed in bits, is the *key length*.
- Let $R = (P - 1)(Q - 1)$ the *totient function*. Note that as $\varphi(N) = \varphi(P) \times \varphi(Q) = (P - 1)(Q - 1)$ since $\varphi(P)$ and $\varphi(Q)$ are $P - 1$ and $Q - 1$ respectively. (Refer to chapter 1 Sec. 1.6 for the definition of *totient function* and other basics of number theory to better understand this section.)
- Choose integer E such that $1 < E < R$ and E is coprime with R . Note that this means $E \in \varphi(R)$. E is released as a part of the public key.
- Determine D such that $ED \bmod R = 1$, that is $D = E^{-1} \bmod R$, the modular multiplicative inverse of $E \bmod R$. D is the private key.

RSA scheme is as follows:

The decryption works as D is chosen such that $ED \bmod R = 1 \implies ED = 1 + xR$ where $x \in \mathbb{Z}$. Hence, we have,

$$\begin{aligned} C^D \bmod N &= (M^E)^D \bmod N \\ &= M^{ED} \bmod N = (M \bmod N)(M^{xR} \bmod N) \\ &= M \bmod N \end{aligned}$$

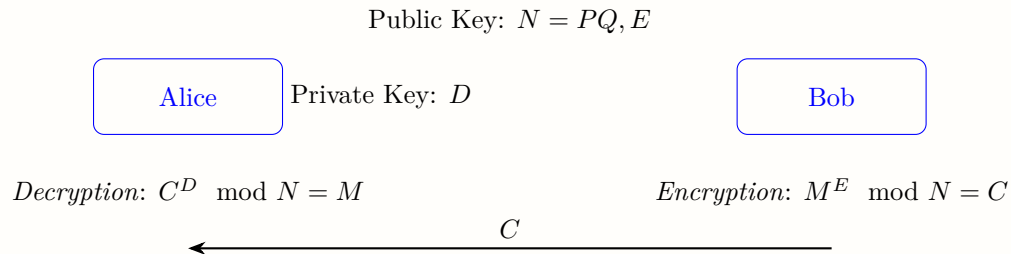


Figure 6.2: RSA

The last line follows from the fact that $M^R \bmod N = 1$ as R is totient of N .

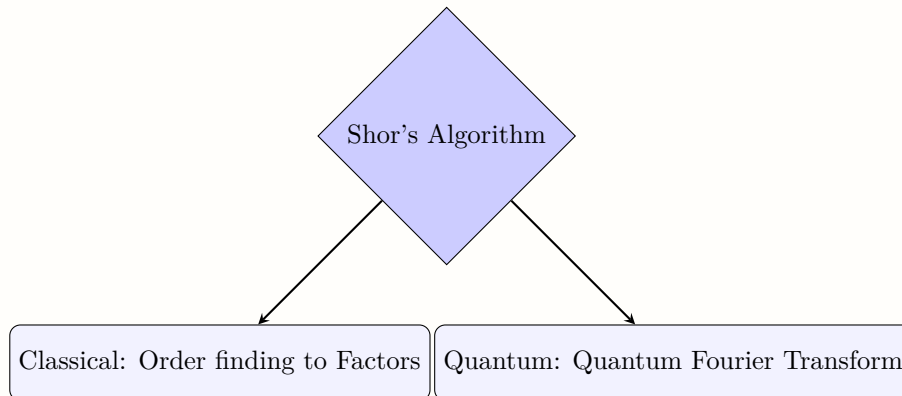
A malicious Eve can eavesdrop on Alice and Bob's conversation and get C . But what guarantees that she can not get M from C given the protocol N and E ?

Classical computers can efficiently compute D such that $ED \bmod R = 1$, provided R is known. So, the real difficulty lies in computing R from E, N , and C , that is, finding the prime factors of N . So, the security of RSA lies in the fact that factoring is a computationally very hard problem. This is no longer true in the case of a quantum computer.

6.2 Overview of Shor's Algorithm

6.2.1 Idea behind Shor's Algorithm

Shor's algorithm consists of a classical and a quantum part.



Shor's algorithm does not allow us to factor a number directly. Instead, it allows us to find the order of an element a modulo n in polynomial time. This, in quantum computers, is done using inverse Quantum Fourier Transform as one of the subroutines.

We will see that finding a factor of n , given the order of some element in $\mathbb{Z}/n\mathbb{Z}$ can be done efficiently even on a classical computer, but no efficient algorithm is known for finding the order of the element.

6.3 Shor's Algorithm

6.3.1 Pseudo-code

INPUT: $N = PQ$ where P and Q are primes

OUTPUT: P, Q

1. Pick a number a that is coprime with N i.e. their gcd is 1.
2. Find the order R of the function $a^R \mod N$.
3. If R is even:
 - Define $x \equiv a^{R/2} \mod N$
 - If $x + 1 \not\equiv 0 \mod N$:
Then the factors P and Q which we are looking for, at least one of them is contained in $\{\gcd(x + 1, N), \gcd(x - 1, N)\}$
4. If either of the above two conditions fails, then pick another a and repeat this all over again.

Remarks. Note that given $a^R \equiv 1 \mod N$ and r is even we can factor $a^R - 1$ as $(a^{\frac{R}{2}} - 1)(a^{\frac{R}{2}} + 1) \equiv 0 \mod N$. If $x \equiv a^{R/2} \mod N$, then possibly either $x - 1$ or $x + 1$ divides N . But note that the former is not possible as we started with the assumption that the orbit of a is of size R , so it can not be $R/2$. If $x + 1$ divides N we just repeat the process again (as said in point 4.)

If both the above fails, then either $x - 1$ or $x + 1$ is a multiple of Q and P , where $N = QP$. Thus finding $\{\gcd(x + 1, N), \gcd(x - 1, N)\}$ gives P and Q .

Example 6.3.1. Consider factoring 15:

1. Let us pick $a = 13$, as 13 is coprime with 15.
2. We need to find the order of $13^x \mod 15$. Since R is the smallest number such that

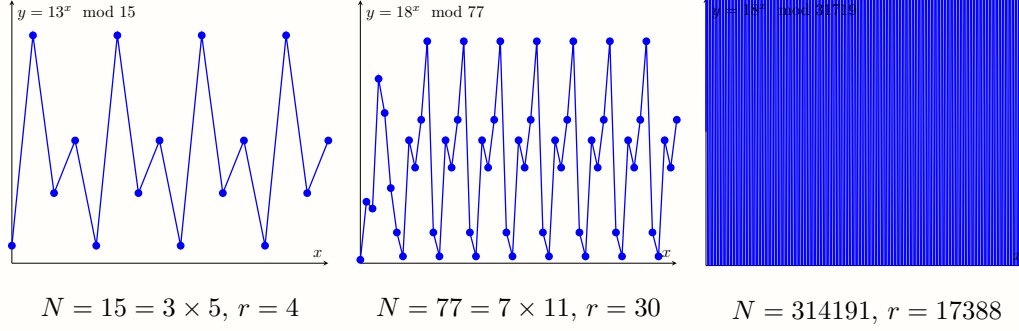
$$\begin{array}{c|ccccccccc} x & 0 & 1 & 2 & 3 & 4 & 5 & 6 & \dots \\ 13^x \mod 15 & \textcircled{1} & 13 & 4 & 7 & \textcircled{1} & 13 & 4 & \dots \end{array}$$

$a^r \equiv 1 \mod N$, here $r = 4$ since the values are periodic about $x = 0, 4, 8, \dots$

3. $R = 4$ is even,
Define $x = a^{R/2} \mod N = 13^{4/2} \mod 15 = 13^2 \mod 15 = 4 \mod 15$.
Therefore, $x \equiv 4 \mod 15$, hence $x + 1 \equiv 4 + 1 \mod 15 \equiv 5 \mod 15 \not\equiv 0 \mod 15$
This implies P or Q is in $\{\gcd(x + 1, N), \gcd(x - 1, N)\}$
Here $\gcd(4 + 1, 15) = 5$, $\gcd(4 - 1, 15) = 3$. So, $P = 5$ and $Q = 3$.

Why can not we implement the above algorithm completely classically?

The reason is that it becomes progressively harder to find the order. We can see this by looking at the plot between $a^z \bmod N$ and z . As the number N grows, the period grows very quickly, and this function appears more and more aperiodic. For $N = 314191$, classical computer runs for about 2 hours in real-time computing. This order-finding part is expedited by using quantum computers.



6.3.2 Classical Part of Shor's Algorithm

In the below section, through Lemma (6.3.1) and Theorem (6.3.3), we will see that, given a composite number n and the order r of some $x \in \mathbb{Z}/n\mathbb{Z}$, we can compute $\gcd(x^{r/2} \pm 1, n)$ efficiently using Euclid's algorithm. This gives a non-trivial factor of n unless r is odd or $x^{r/2} \equiv -1 \pmod{n}$. In particular, if n is a semi-prime, i.e., it is a product of two primes p and q , then Theorem (6.3.3) implies that n will be factored with probability $\frac{1}{2}$.

6.3.2.1 Factoring as Order finding

We will show that the problem of finding a non-trivial factor to n can be reduced (efficiently) to finding the order of a non-trivial element in $\mathbb{Z}/n\mathbb{Z}$.

Lemma 6.3.1. *Given a composite number n , and x non-trivial square root of 1 modulo n , i.e. $x^2 \equiv 1 \pmod{n}$ but x is neither 1 nor $-1 \pmod{n}$, then either $\gcd(x-1, n)$ or $\gcd(x+1, n)$ is a non-trivial factor of n .*

Proof. Since $x^2 \equiv 1 \pmod{n}$, we have $x^2 - 1 \equiv 0 \pmod{n}$. Factoring, we get $(x-1)(x+1) \equiv 0 \pmod{n}$. This implies that n is a factor of $(x+1)(x-1)$. Since $(x \pm 1) \not\equiv 0 \pmod{n}$, n has a non-trivial factor with $x+1$ or $x-1$. To find this common factor efficiently, we apply Euclid's algorithm to get $\gcd(x-1, n)$ or $\gcd(x+1, n)$. ■

Example 6.3.2. *Let $n = 55 = 5 \times 11$. We find that 34 is a square root of 1 mod n since $34^2 = 1156 = 1 + 21 \times 55$. Computing, we get $\gcd(33, 55) = 11$ and $\gcd(35, 55) = 5$.*

Lemma 6.3.2. *Let n be odd, then at least half the elements in $(\mathbb{Z}/n\mathbb{Z})^\times$ have even order.*

Proof. Suppose $\text{ord}(x) = r$ is odd. Then $(-x)^r = (-1)^r x^r = (-1)^r = -1 \pmod{n}$. Hence, $-x$ must have order $2r$, which is even. Therefore, at least half the elements in $(\mathbb{Z}/n\mathbb{Z})^\times$ have even order. ■

Equipped with these tools, we will proceed to prove the main result that allows us to reduce the factorisation of n to find the order of an element in $\mathbb{Z}/n\mathbb{Z}$.

Theorem 6.3.3. *Let n be an odd integer and let $n = p_1^{e_1} p_2^{e_2} \cdots p_k^{e_k}$ be the prime factorization of n . Then the probability that a uniformly randomly chosen $x \in \mathbb{Z}/n\mathbb{Z}$ has even order r and $x^{r/2} \not\equiv -1 \pmod{n}$ is at least $1 - \frac{1}{2^{k-1}}$.*

Proof. By the Chinese Remainder Theorem, choosing $x \in (\mathbb{Z}/n\mathbb{Z})^\times$ (uniform) randomly is equivalent to choosing $x_i \in (\mathbb{Z}/p_i^{e_i}\mathbb{Z})^\times$ for each p_i randomly. Let r be the order of x and let r_i be the order of x_i . In particular, $x^{r/2}$ is never $1 \pmod{n}$. We want to show that the probability of either r being odd or $x^{r/2} \equiv -1 \pmod{n}$ is at most $\frac{1}{2^{k-1}}$.

Note that $r = \text{lcm}(r_1, r_2, \dots, r_k)$ (where lcm denotes the least common multiple). To see this, $x^r \equiv 1 \pmod{n}$, $x^r \equiv 1 \pmod{p_i^{e_i}}$, hence r is a multiple of each r_i . It is the least such number and hence the least common multiple of the r_i 's.

Suppose that r is odd. This happens only if all of the r_i 's are odd. r_i is odd with probability at most one-half by Lemma (6.3.2). Hence, r is odd with probability at most $\frac{1}{2^k}$.

Now, suppose that r is even. We still have to worry about the possibility that $x^{r/2} \equiv \pm 1 \pmod{n}$. By the Chinese Remainder Theorem, this happens only if $x^{r/2} \equiv \pm 1 \pmod{p_i^{e_i}}$ for every p_i . We need to avoid these cases since $\equiv +1$ means r wasn't the order, and $\equiv -1$ doesn't yield a useful factorisation. The probability of choosing an x such that one of these two cases happens is $2 \cdot 2^{-k} = 2^{-k+1}$.

Combining the probabilities, we get a success probability of at least $(1 - 2^{-k})(1 - 2^{-k+1}) \geq 1 - 3 \cdot 2^{-k}$. ■

By Lemma (6.3.1) and Theorem (6.3.3), given a composite number n and the order r of some $x \in \mathbb{Z}/n\mathbb{Z}$, we can compute $\gcd(x^{r/2} \pm 1, n)$ efficiently using Euclid's algorithm. This gives a non-trivial factor of n unless r is odd or $x^{r/2} \equiv -1 \pmod{n}$. In particular, if n is a semi-prime, i.e., it is a product of two primes p and q , then Theorem (6.3.3) implies that n will be factored with probability $\frac{1}{2}$.

6.3.3 Quantum part of Shor's Algorithm

6.3.3.1 Discrete Fourier Transform (DFT)

Let's start with a familiar idea. Imagine you're listening to a piece of music. The music is made up of different notes (frequencies) that together create a melody. Now, if you wanted to analyze which notes are present, you'd try to pick apart the sound into its individual frequencies. This is essentially what the Fourier transform does by breaking down a complex signal into a sum of simple sinusoidal waves, each with its own frequency, amplitude, and phase.

In the classical setting, when we have a periodic function, say, one that repeats every T units, the Fourier transform will show us spikes at specific frequencies. The most prominent spike is at the fundamental frequency, which is $f_0 = \frac{1}{T}$. This is the basic beat of the function, the frequency at which the pattern repeats. But a typical periodic function isn't just a simple sine wave, and might be a more complex shape. This complexity is reflected in the presence of harmonics, which are spikes at frequencies that are integer multiples of

the fundamental frequency (i.e., $2f_0, 3f_0, \dots$).

In practice, especially when working with digital data, we use the Discrete Fourier Transform (DFT). The DFT algorithm takes a sequence of data points (samples of our function) and computes how much of each frequency is present in the signal. When you run a DFT on a periodic function, you see peaks in the output at the frequencies where the function has a strong periodic component.

Here is an illustration for $y = \sin(2\pi\nu x)$, whose Fourier transform DFT \tilde{y} has the peak at ν . Note that the broadening of the unique peak occurs due to the finiteness of the data size. There is a single peak since $\sin(2\pi\nu x)$ has the fundamental mode and no additional harmonics.

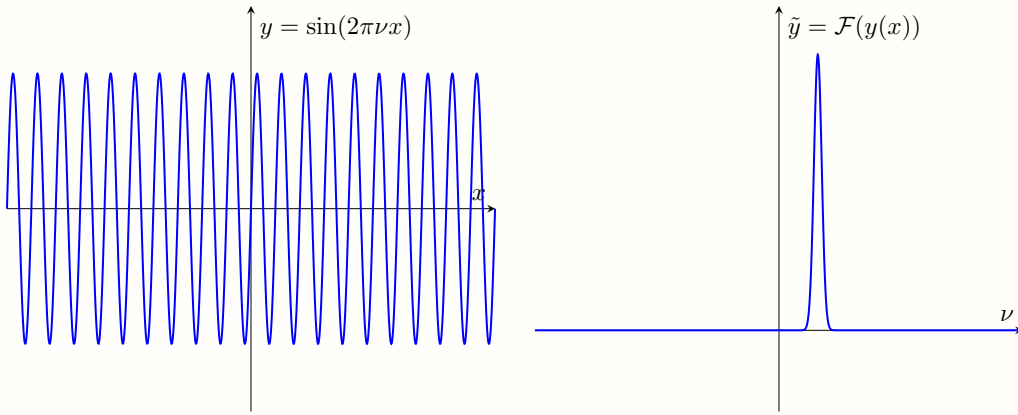


Figure 6.3: Discrete Fourier Transform of sin function

Now consider the function $f(x) = a^x \bmod N$ where $a \in \mathbb{Z}$ and $N \in \mathbb{N}$, which is periodic over the scales of N . Decomposing as a DFT, we have the following interpretation.

We note that the peaks of the DFT correspond to the Fourier fundamental frequencies, which are integral multiples of the period.

We generalize this idea of hunting for fundamental frequencies to a general vector by describing the DFT as the tool that decomposes a vector of complex numbers into its intrinsic frequency components. The formal definition of the algorithm is as follows:

INPUT: A vector of complex numbers x_0, x_1, \dots, x_{N-1} , where N is a fixed parameter (assuming $N = 2^n$).

OUTPUT: A vector of complex numbers y_0, y_1, \dots, y_{N-1} , such that

$$y_k = \frac{1}{\sqrt{N}} \sum_{j=0}^{N-1} e^{2\pi i j k / N} x_j.$$

Let's build up the picture step by step. The DFT decomposes the input vector into a linear

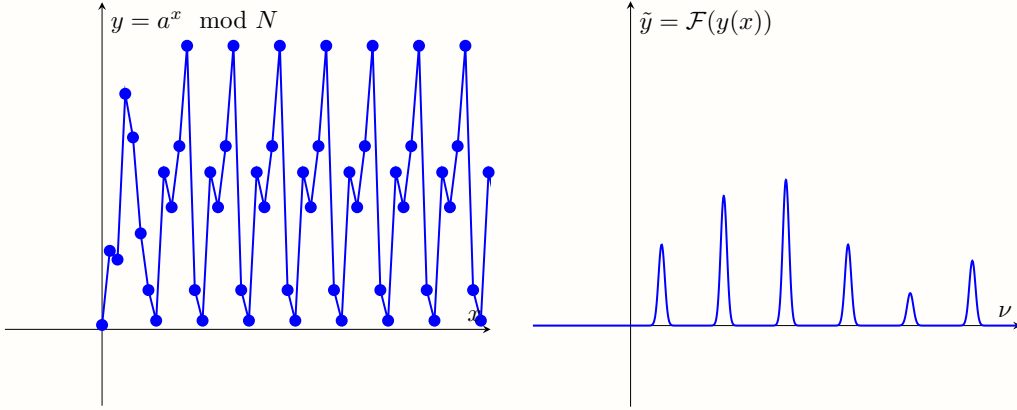


Figure 6.4: Discrete Fourier Transform of $f(x) = a^x \bmod N$

combination of complex exponentials. These exponentials, given by the factors $e^{2\pi ijk/N}$, serve as basis functions that oscillate at specific frequencies. For each k , we can think of $e^{2\pi ik/N}$ as a complex vector with N entries, individually by

$$v_k = \left(\frac{1}{N}, \frac{1}{N}e^{2\pi ik/N}, \frac{1}{N}e^{4\pi ik/N}, \dots, \frac{1}{N}e^{2\pi ik(N-1)/N} \right).$$

These N vectors form an orthonormal basis of \mathbb{C}^N , and can be used to decompose any vector into components along these vectors. We can directly compute the dot product of our vector to note the component along the suitable basis vector. When the input signal has a periodic structure, these basis vectors align with the natural periodicities of the signal, producing prominent peaks in the output. The term $e^{2\pi ijk/N}$ can be viewed as a rotating phase factor. For a fixed k , as j runs over the values 0 to $N-1$, these exponentials trace out a complete cycle. When the input signal x_j resonates with this cycle (that is, when the signal contains a frequency component matching k/N), the sum in Equation (6.3.3.1) reinforces this frequency component, leading to a peak in the output y_k .

Remarks. Using the Fast Fourier Transform algorithm (FFT)¹, we can do DFT faster. We will see that during the exercise of making a Quantum Fourier Transform algorithm, FFT will appear as a byproduct.

6.3.4 Quantum Fourier Transform (QFT)

Similar to DFT definition, QFT on an orthonormal basis $|0\rangle \dots |N-1\rangle$ is defined to be a linear operator with the following action on the basis states,

¹Check out the YouTube video [The Most Important Algorithm Of All Time by Veritasium](#) for an intuitive discussion of the algorithm.

$$QFT|j\rangle = \frac{1}{\sqrt{2^n}} \sum_{k=0}^{2^n-1} e^{2\pi i j k / 2^n} |k\rangle.$$

The action on an arbitrary state may be written as

$$\sum_{j=0}^{N-1} x_j |j\rangle = \sum_{k=0}^{N-1} y_k |k\rangle,$$

where y_k are DFT of the amplitudes x_j . This expression strengthens our understanding of the basis transformation nature of the algorithm.

Remarks. *It is not obvious from the definition of QFT, but this transformation is a unitary transformation and thus can be implemented as the dynamics for a quantum computer. The theorem below will show this fact and also give an expression that can be easily interpreted when designing a quantum circuit for the QFT algorithm.*

Can you prove QFT as defined above is unitary?

$$U|j\rangle = \frac{1}{\sqrt{N}} \sum_{k=0}^{N-1} e^{2\pi i j \cdot k / N} |k\rangle \quad \text{QFT Claim: } U \text{ is Unitary}$$

$$\begin{aligned} U^\dagger U &= \frac{1}{N} \left(\sum_{k=0}^{N-1} e^{-2\pi i j \cdot k / N} \langle k| \right) \left(\sum_{k'=0}^{N-1} e^{2\pi i j' \cdot k' / N} |k'\rangle \right) \\ &= \frac{1}{N} \sum_{k=0}^{N-1} \sum_{k'=0}^{N-1} e^{\frac{2\pi i j j'}{N} (k' - k)} \langle k | k' \rangle \\ &= \frac{1}{N} \sum_{k=0}^{N-1} \mathbb{I} \cdot 1 = \frac{N}{N} \mathbb{I} = \mathbb{I} \end{aligned}$$

We consider $N = 2^n, n \in \mathbb{Z}$. As earlier, the basis $|0\rangle \dots |N-1\rangle$ is $|0\rangle \dots |2^n-1\rangle$ thus can be represented using n bit-string. Thus, $|j\rangle = |j_1 \dots j_n\rangle$ where $j = j_1 2^{n-1} + j_2 2^{n-2} + \dots + j_n 2^0$. Also $0.j_1 j_2 \dots j_m$ is binary fraction $\frac{j_1}{2} + \frac{j_2}{2^2} + \dots + \frac{j_m}{2^{m-1+1}}$.

Theorem 6.3.4. (*QFT Representation*)

$$|j_1, \dots, j_n\rangle \xrightarrow{QFT} \frac{1}{2^{n/2}} [(|0\rangle + e^{2\pi i 0 \cdot j_n} |1\rangle) (|0\rangle + e^{2\pi i 0 \cdot j_{n-1} j_n} |1\rangle) \dots (|0\rangle + e^{2\pi i 0 \cdot j_1 j_2 \dots j_n} |1\rangle)].$$

Proof.

$$\begin{aligned} QFT|j\rangle &= \frac{1}{2^{n/2}} \sum_{k=0}^{2^n-1} e^{\frac{2\pi i j k}{2^n}} |k\rangle = \frac{1}{2^{n/2}} \sum_{k_1=0}^1 \dots \sum_{k_n=0}^1 e^{\frac{2\pi i j (k_1 2^{n-1} + k_2 2^{n-2} + \dots + k_n 2^0)}{2^n}} |k\rangle \\ &= \frac{1}{2^{n/2}} \sum_{k_1=0}^1 \dots \sum_{k_n=0}^1 e^{2\pi i j (k_1 2^{-1} + k_2 2^{-2} + \dots + k_n 2^{-n})} |k\rangle \end{aligned}$$

$$\begin{aligned}
&= \frac{1}{2^{n/2}} \sum_{k_1=0}^1 \cdots \sum_{k_n=0}^1 e^{2\pi i j \sum_{l=1}^n k_l 2^{-l}} |k_1 \cdots k_n\rangle \\
&= \frac{1}{2^{n/2}} \sum_{k_1=0}^1 \cdots \sum_{k_n=0}^1 \bigotimes_{l=1}^n e^{2\pi i j k_l 2^{-l}} |k_l\rangle \\
&= \frac{1}{2^{n/2}} \bigotimes_{l=1}^n \left[\sum_{k_l=0}^1 e^{2\pi i j k_l 2^{-l}} |k_l\rangle \right] \\
&= \frac{1}{2^{n/2}} \bigotimes_{l=1}^n (|0\rangle + e^{2\pi i j 2^{-l}} |1\rangle) \\
&= \frac{1}{2^{n/2}} [|0\rangle + e^{2\pi i 0 \cdot j_n} |1\rangle] (|0\rangle + e^{2\pi i 0 \cdot j_{n-1} j_n} |1\rangle) \cdots (|0\rangle + e^{2\pi i 0 \cdot j_1 j_2 \cdots j_n} |1\rangle)
\end{aligned}$$

■

Is there a relation between Hadamard operator and QFT?

Consider $U|00 \dots 0\rangle$, where $U \rightarrow \text{QFT}$

$$\begin{aligned}
&= \frac{1}{\sqrt{N}} \sum_{k=0}^{N-1} e^{2\pi i j k / N} |k\rangle \\
&= \frac{1}{\sqrt{N}} \sum_{k=0}^{N-1} 1 \cdot |k\rangle
\end{aligned}$$

The coefficients have become 1 as $j \cdot k = j_1 \cdot k_1 + j_2 \cdot k_2 \cdots j_N \cdot k_N = 0$

So, $U|00 \dots 0\rangle = \frac{1}{\sqrt{N}} \sum_{k=0}^{N-1} |k\rangle$ which is the equal superposition of all basis, which is nothing but Hadamard on $|00 \dots 0\rangle$.

6.3.4.1 Quantum circuit for implementing QFT

Let the gate R_k denote the unitary transformation, $R_k \equiv \begin{pmatrix} 1 & 0 \\ 0 & e^{\frac{2\pi i}{2^k}} \end{pmatrix}$.

To see that the pictured circuit Fig. 6.5, computes the quantum Fourier transform, consider what happens when the state $|j_1 \cdots j_n\rangle$ is input. Applying the Hadamard gate to the first bit produces the state

$$\frac{1}{\sqrt{2}} (|0\rangle + e^{2\pi i 0 \cdot j_1} |1\rangle) |j_2 \cdots j_n\rangle,$$

since $e^{2\pi i 0 \cdot j_1} = -1$ when $j_1 = 1$, and is $+1$ otherwise. Applying the controlled R_2 gate produces the state,

$$\frac{1}{\sqrt{2}} (|0\rangle + e^{2\pi i 0 \cdot j_1 j_2} |1\rangle) |j_2 \cdots j_n\rangle.$$

We continue applying the controlled R_3, R_4 through R_n gates, each of which adds an extra bit to the phase of the coefficient of the first $|1\rangle$. At the end of this procedure, we have the state,

$$\frac{1}{\sqrt{2}} (|0\rangle + e^{2\pi i 0.j_1 j_2 \dots j_n} |1\rangle) |j_2 \dots j_n\rangle.$$

Next, we perform a similar procedure on the second qubit. The Hadamard gate puts us in the state,

$$\frac{1}{\sqrt{2^2}} (|0\rangle + e^{2\pi i 0.j_1 j_2 \dots j_n} |1\rangle) (|0\rangle + e^{2\pi i 0.j_2} |1\rangle) |j_3 \dots j_n\rangle,$$

and the controlled- R_2 through R_{n-1} gates yield the state,

$$\frac{1}{\sqrt{2^2}} (|0\rangle + e^{2\pi i 0.j_1 j_2 \dots j_n} |1\rangle) (|0\rangle + e^{2\pi i 0.j_2 \dots j_n} |1\rangle) |j_3 \dots j_n\rangle.$$

We continue in this fashion for each qubit, giving a final state,

$$\frac{1}{\sqrt{2^n}} (|0\rangle + e^{2\pi i 0.j_1 j_2 \dots j_n} |1\rangle) (|0\rangle + e^{2\pi i 0.j_2 \dots j_n} |1\rangle) \dots (|0\rangle + e^{2\pi i 0.j_n} |1\rangle).$$

Swap operations (which are not shown in the figure) are then used to reverse the order of the qubits, which are simple transmutations of the elements leading to a reverse permutation. After the swap operations, the state of the qubits is,

$$\frac{1}{\sqrt{2^n}} (|0\rangle + e^{2\pi i 0.j_n} |1\rangle) (|0\rangle + e^{2\pi i 0.j_{n-1} j_n} |1\rangle) \dots (|0\rangle + e^{2\pi i 0.j_1 j_2 \dots j_n} |1\rangle).$$

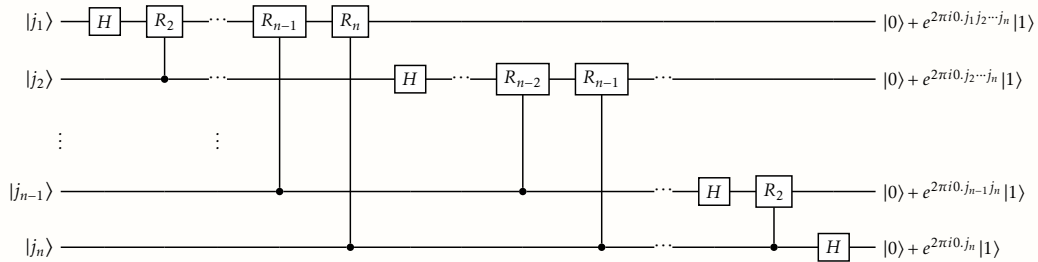


Figure 6.5: Quantum Circuit for QFT

Comparing with Equation (6.3.4), we see that this is the desired output from the quantum Fourier transform. This construction also proves that the quantum Fourier transform is

unitary since each gate in the circuit is unitary.

How many gates does this circuit use? We start by doing a Hadamard gate and $n - 1$ conditional rotations on the first qubit which is a total of n gates. This is followed by a Hadamard gate and $n - 2$ conditional rotations on the second qubit, for a total of $n + (n - 1)$ gates. Continuing in this way, we see that $n + (n - 1) + \dots + 1 = \frac{n(n+1)}{2}$ gates are required, plus the gates involved in the swaps. At most $\frac{n}{2}$ swaps are required, and each swap can be accomplished using three controlled- X gates. Therefore, this circuit provides a $\mathcal{O}(n^2)$ algorithm for performing the quantum Fourier transform.

In contrast, the best classical algorithms for computing the discrete Fourier transform on 2^n elements are algorithms such as the Fast Fourier Transform (FFT), which compute the discrete Fourier transform using $\mathcal{O}(n2^n)$ gates. That is, it requires exponentially more operations to compute the Fourier transform on a classical computer than it does to implement the quantum Fourier transform on a quantum computer.

6.3.4.2 FFT from QFT

DFT takes $\Theta(2^{2n})$ operations on an input with $2n$ components. This is quite easy to see if we look at the $2^n \times 2^n = 2^{2n}$ matrix of DFT:

$$W = \frac{1}{\sqrt{2n}} \begin{pmatrix} 1 & 1 & 1 & 1 & \dots & 1 \\ 1 & \omega & \omega^2 & \omega^3 & \dots & \omega^{2n-1} \\ 1 & \omega^2 & \omega^4 & \omega^6 & \dots & \omega^{2(2n-1)} \\ 1 & \omega^3 & \omega^6 & \omega^9 & \dots & \omega^{3(2n-1)} \\ \vdots & \vdots & \vdots & \vdots & \ddots & \vdots \\ 1 & \omega^{2n-1} & \omega^{2(2n-1)} & \omega^{3(2n-1)} & \dots & \omega^{(2n-1)(2n-1)} \end{pmatrix}.$$

If we multiply W with a vector and count the operations, we get the result.

Equation (6.3.4) allows you to take advantage of the fact that the Fourier transformed $|j_1, j_2, \dots, j_n\rangle$ is made out of n tensored 2×1 vectors. So, we process each 2×1 vector independently by performing the following n mappings:

$$\frac{1}{\sqrt{2}} \begin{pmatrix} |0\rangle + |1\rangle \\ \vdots \\ |0\rangle + |1\rangle \end{pmatrix} \rightarrow \frac{1}{\sqrt{2}} \begin{pmatrix} |0\rangle + e^{2\pi i 0 \cdot j_n} |1\rangle \\ \vdots \\ |0\rangle + e^{2\pi i 0 \cdot j_1 \dots j_n} |1\rangle \end{pmatrix}.$$

Each mapping takes a constant number of operations in n as it is simply multiplying a 2×1 vector by a 2×2 phase matrix.

$$R_k = \begin{pmatrix} 1 & 0 \\ 0 & e^{\frac{2\pi i}{2^k}} \end{pmatrix}.$$

Hence, we perform n matrix-vector multiplication to process a single $|j_1 \dots j_n\rangle$.

We know that an arbitrary vector $|\psi\rangle$ on n qubits can be written as a linear combination of 2^n binary kets $|j_1, j_2, \dots, j_n\rangle$. For example, for $n = 2$, an arbitrary state can be written as a linear combination of 2^2 binary kets as follows:

$$|\psi\rangle = a|00\rangle + b|01\rangle + c|10\rangle + d|11\rangle.$$

Therefore, to transform $|\psi\rangle$ on n qubits, we need to process 2^n binary vectors $|j_1, \dots, j_n\rangle$ by performing n mappings described above. Since each such binary vector requires n matrix-vector multiplications, and there are 2^n of them, it takes $\Theta(n2^n)$ operations.

6.3.5 QFT in Shor's algorithm

For the following section, we will assume that N' is a composite odd integer which is not a power of prime (the algorithm fails otherwise). If N' is even, we can just factor out all the powers of 2 until we get an odd integer, then run the algorithm on the resulting integer. We can test whether N' is a prime efficiently using classical primality tests such as the AKS test and the Miller-Rabin test ². We can also test if N' is a power of prime efficiently by taking the k^{th} root of N' until $\sqrt[k]{N'} < 2$.

Given N' , we choose $N = 2^n$ such that $N' < N < 2N'$ (i.e., choose the unique power of 2 in that range). We will be working with two registers (two arrays of qubits), such that each of them holds n qubits. At first, the registers are $|0\rangle \otimes |0\rangle$.

We put the first register in the uniform superposition of numbers $x \bmod N$ by using the QFT (This is equivalent to applying Hadamard gate to all qubits in the first register),

$$|0\rangle \xrightarrow{QFT} \frac{1}{\sqrt{N}} \sum_{x=0}^{N-1} |x\rangle \otimes |0\rangle.$$

Now suppose $f(x) = a^x \bmod N$. Note that the period of f is the same as the order of a , given by r . Given some base a , can we compute $f(x)$ efficiently? The answer is yes; we can just exponentiate by repeated squaring!

We need to apply f to the contents of the first register and store the result of $f(x)$ in the second register. To do so, we can construct f as a quantum function. It turns out that this is the bottleneck of the algorithm since implementing f on a quantum computer requires a lot of quantum gates³. Still, Shor's algorithm is much faster than factoring on a classical computer.

We have the state $\frac{1}{N} \sum_{x=0}^{N-1} |x\rangle \otimes |f(x)\rangle$. Apply the inverse QFT to the first register, and we get

$$QFT^{-1} \frac{1}{\sqrt{N}} \sum_{x=0}^{N-1} |x\rangle \otimes |f(x)\rangle = \frac{1}{\sqrt{N}} \sum_{x=0}^{N-1} (QFT^{-1}|x\rangle) \otimes |f(x)\rangle = \frac{1}{N} \sum_{x,y=0}^{N-1} e^{-\frac{2\pi i xy}{N}} |y\rangle \otimes |f(x)\rangle$$

²Refer to the phenomenal paper *Primes in P* by our fellow Indians [Manindra, Agrawal and Neeraj, Kayal and Nitin, Saxena \[2002\]](#).

³Refer to Shor's paper [Shor \[1997\]](#).

Remarks. Note that inverse QFT is equivalent to QFT^\dagger , which is the case for every quantum gate.

Measure the second register, then after applying inverse QFT, measure the first register. Depending on the value do classical processing, as mentioned in Sec. 6.3.1.

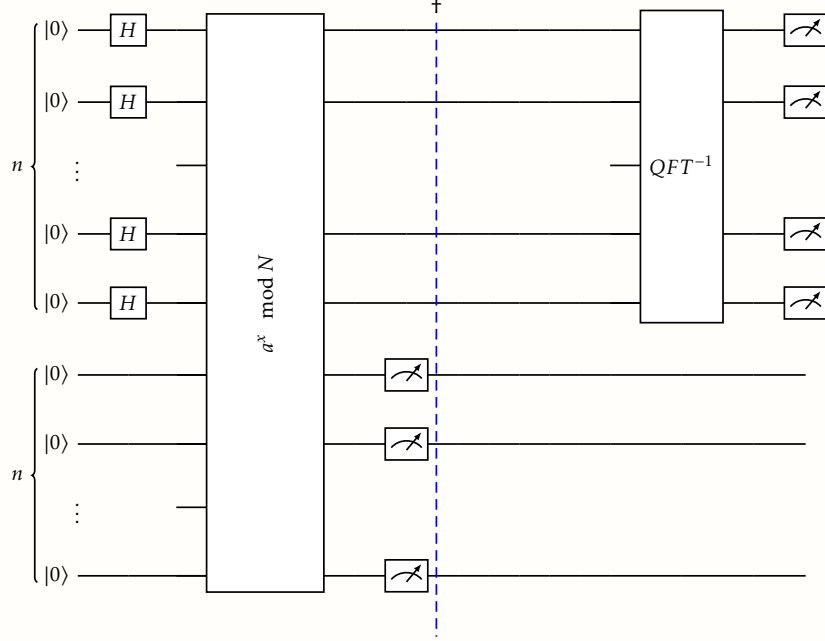


Figure 6.6: Quantum Circuit for Shor's Algorithm

Example 6.3.3. Again consider the number 15 ($|1111\rangle$ in 4 qubits representation). This time we will use the circuit to factor the number.

1. Start with set of 2 registers at the state $|0\rangle^{\otimes 4} |0\rangle^{\otimes 4}$.
2. Now apply Hadamard on the first set of register,

$$\left[H^{\otimes 4} |0\rangle^{\otimes 4} \right] |0\rangle^{\otimes 4} = \frac{1}{4} [|0\rangle + |1\rangle + \dots + |15\rangle] |0\rangle^{\otimes 4}.$$

Here the numbers inside ket are in base 10 representation. In base 2, they are all possible 4 bitstrings.

3. Applying $f(x)$ on the second register

$$= \frac{1}{4} [|0\rangle |0 \oplus 13^0 \mod 15\rangle + |0\rangle |0 \oplus 13^1 \mod 15\rangle + \dots].$$

Note that $0 \oplus$ (i.e. XOR) something is the number itself

$$= \frac{1}{4} \left[|0\rangle |1\rangle + |1\rangle |13\rangle + |2\rangle |4\rangle + |3\rangle |7\rangle + \right. \quad (6.1)$$

$$|4\rangle |1\rangle + |5\rangle |13\rangle + |6\rangle |4\rangle + |7\rangle |7\rangle + \quad (6.2)$$

$$|8\rangle |1\rangle + |9\rangle |13\rangle + |10\rangle |4\rangle + |11\rangle |7\rangle + \quad (6.3)$$

$$\left. |12\rangle |1\rangle + |13\rangle |13\rangle + |14\rangle |4\rangle + |15\rangle |7\rangle \right]. \quad (6.4)$$

$$(6.5)$$

4. We now measure the second register (This measurement happens before applying inverse QFT)

Suppose after measuring second register, we get $|7\rangle$. Implies, we have the superposition $\frac{1}{2} [|3\rangle + |7\rangle + |11\rangle + |15\rangle] \otimes |7\rangle$. Note the normalisation, $\frac{1}{2}$, i.e. probabilities have changed.

5. Now apply inverse QFT (Equation (6.3.5)) to the first register.

If we apply and compute, we will find that phases will interfere and cancel out. The only terms which will remain are

$$= \frac{1}{8} [4|0\rangle + 4i|4\rangle + 4|8\rangle + 4i|12\rangle].$$

6. The final step is to measure the first register.

We will get $|0\rangle, |4\rangle, |8\rangle$ or $|12\rangle$ with equal probability of $\frac{1}{4}$.

We have completed the quantum part of Shor's algorithm. After this, all that is left is doing the classical post-processing. The measurement results peak near $j \times \frac{N}{R}$ for some integer $j \in \mathbb{Z}$.

Analysing the measurement results:

- $|0\rangle$ is trivial. If we measure $|0\rangle$, restart.
- $|4\rangle$ $j^{16/R} = 4$ One possibility (the lowest one) is $j = 1$
Implies $R = 4$ even, which is good.
 $x = a^{R/2} \mod N = 13^{4/2} \mod 15 = 13^2 \mod 15 = 4 \mod 15$.
Therefore, $x \equiv 4 \mod 15$ and $x + 1 \equiv 4 + 1 \mod 15 \equiv 5 \mod 15 \not\equiv 0 \mod 15$
Thereby, P or Q is in $\{\gcd(x + 1, N), \gcd(x - 1, N)\}$
Here $\gcd(4 + 1, 15), \gcd(4 - 1, 15) = 5, 3$. So, $P = 5$ and $Q = 3$.
- For $|8\rangle$ and $|12\rangle$, we get one of the factors, and the algebra works just like above.

Remarks. Note that the above phase cancellations were possible because of interference which is a quantum phenomenon. This enables a drastic reduction of terms, thus giving an exponential speed-up compared to classical computers.

It is known that if we repeat the above algorithm $\mathcal{O}(\log \log(n))$ times and almost guarantee that we find R^4 .

⁴This non-trivial calculation can be found out in great detail in the book [Nielsen and Chuang \[2011\]](#).

6.4 How complex is Shor's Algorithm?

The bottleneck in the quantum factoring algorithm, i.e., the piece of the factoring algorithm that consumes the most time and space, is computing the function $f(x) = a^x \bmod N$ modular exponentiation. The modular exponentiation problem is, given N , x , and r , find $x^r \bmod N$. The best classical method for doing this is to repeatedly square $x \bmod N$ to get $x^{2^m} \bmod N$ for $m \leq \log_2 r$, and then multiply a subset of these powers ($\bmod N$) to get $x^r \bmod N$. If we are working with n -bit numbers, this requires $\mathcal{O}(n)$ squaring and multiplications of n -bit numbers ($\bmod N$). Asymptotically, the best classical result for gate arrays for multiplication is the Schönhage–Strassen⁵ algorithm. This gives a gate array for integer multiplication that uses $\mathcal{O}(n \log n \log \log n)$ gates to multiply two n -bit numbers. Thus, asymptotically, modular exponentiation requires $\mathcal{O}(n^2 \log n \log \log n)$ time. Making this reversible would naively cost the same amount in space⁶. However, one can reuse the space used in the repeated squaring part of the algorithm and thus reduce the amount of space needed to essentially require for multiplying two n -bit numbers. Thus, modular exponentiation can be done in $\mathcal{O}(n^2 \log n \log \log n)$ time and $\mathcal{O}(n \log n \log \log n)$ space.

As seen earlier, QFT is $\mathcal{O}(n^2)$ and repeating the above algorithm 6.3.1 $\mathcal{O}(\log \log(n))$ times can almost guarantee that we find r . Overall the time complexity of Shor's algorithm is $\mathcal{O}(n^2 \log n \log \log n)$, which is exponential speed up compared to all classically known algorithms!

6.5 Quantum Phase Estimation

PROBLEM STATEMENT: Given a unitary operator U has an eigenvector $|u\rangle$ with eigenvalue $e^{2\pi i\phi}$, where the value of ϕ is unknown. The goal of the phase estimation algorithm is to estimate ϕ .

Note that given a unitary U ($UU^* = I$); we know that all its eigenvalues have norm 1. Since any complex number can be written as $re^{2\pi i\theta}$, all eigenvalues of U should be of the form $e^{2\pi i\theta}$ for some θ . To determine the eigenvalue, it is enough to find this θ , called the phase of the eigenvalue or eigenbasis.

Quantum phase estimation is a useful subroutine in quantum computing that uses quantum Fourier transform. Suppose we have black boxes capable of preparing the state $|u\rangle$ and performing the controlled- U^{2^j} operation for suitable non-negative integers j .

The phase estimation subroutine, given a unitary U and its eigenvector $|u\rangle$, finds the phase of the eigenvalue corresponding to the eigenvector $|u\rangle$. To be precise, the algorithm will take the eigenvector $|u\rangle$ as input, and it needs the ability to perform controlled U^{2^i} ($i \leq k$) operations; using those, it determines the corresponding eigenphase.

⁵Refer to [Schönhage and Strassen \[1971\]](#).

⁶Refer to [Shor \[1997\]](#) for more details

To start with, we will also assume that we have the ability to perform U^l for all $l \leq 2^k = n$ (instead of just controlled U^{2^k}). Later we will show that controlled U^{2^i} ($i \leq k$) operators can be used to perform U^l for all $l \leq 2^k$.

We will start with the state $|0, u\rangle$, where the first part of the register holds k qubits and second register holds the eigenvector $|u\rangle$. Then we will apply Hadamard on the first part and obtain,

$$\frac{1}{2^{k/2}} \sum_{l=1}^{2^k} |l, u\rangle$$

Now we can perform the operation $|l, u\rangle \rightarrow |l\rangle U^l |u\rangle$. Notice that this can be done classically on the basis states and hence can be done quantumly.

This gives us the state,

$$\frac{1}{2^{k/2}} \sum_{l=1}^{2^k} |l\rangle U^l |u\rangle = \frac{1}{2^{k/2}} \sum_{l=1}^{2^k} e^{2\pi i \theta l} |l, u\rangle.$$

Some thought shows that the first part of the register is the Fourier transform of $2^k \theta$. Hence applying inverse Fourier transform, we get the state $|2^k \theta\rangle$.

$$\frac{1}{2^{t/2}} \sum_{j=0}^{2^t-1} e^{2\pi i \varphi j} |j\rangle |u\rangle \rightarrow |\tilde{\varphi}\rangle |u\rangle$$

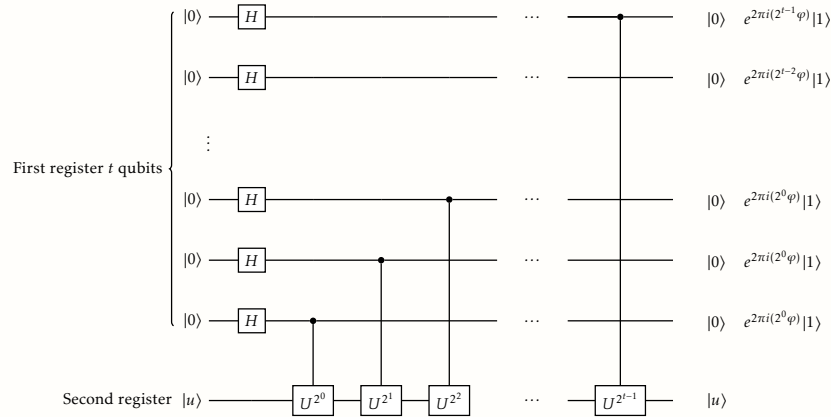


Figure 6.7: Application of controlled- U^k

If we are only given the controlled versions of U^{2^l} where $l \leq k$, then how can we achieve the same phase estimation? Notice that l now varies only up to k . Essentially, we are given the

power to apply $U, U^2, U^4, \dots, U^{2^k}$.

The simple idea is to break any integer $0 \leq h \leq 2^k$ as powers of 2. Then using the controlled version, we can apply U^h .

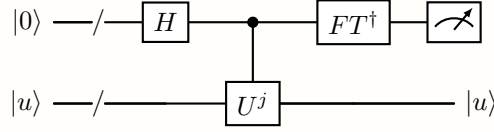


Figure 6.8: Quantum Phase Estimation

Let us see how to take care of the assumptions we made, there are only k bits in the expansion of θ and we have the eigenvector as a quantum state $|u\rangle$.

Most of the time, it is not possible to know the number of digits in the binary expansion of θ beforehand. What can be done in this case? If we want to approximate θ up to k bits of accuracy, using the same circuit with $k + f(\epsilon)$ qubits instead of k qubits will give us the answer with probability $1 - \epsilon$. Here, ϵ should be treated as a parameter, and $f(\epsilon)$ is some function of ϵ .

Suppose we don't have the eigenvector $|u\rangle$. If the same procedure is done over $|\psi\rangle = \sum_i \alpha_i |u_i\rangle$, we will get the phase corresponding to $|u_i\rangle$ with probability $|\alpha_i|^2$.

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Chapter 7

Grover's Search Algorithm

"There are only two tragedies in life: one is not getting what one wants, and the other is getting it."

– Oscar Wilde, *Lady Windermere's Fan*

7.1 Introduction

The problem of searching through unstructured data is ubiquitous, and any improvement will help a lot of applications. If it were structured data, like given an ordered list, we could exploit the order and search faster (like using binary search). But when there is no order in the list, we have no other option, in a classical computer, other than going through all elements one by one, making $\Theta(n)$ queries to see each element. In quantum computing, we can do it in $\mathcal{O}(\sqrt{n})$ queries¹. When the data set is huge, this quadratic 'speed-up' over its classical analogue can save a lot of time. The genius behind the quantum search algorithm is Lov Kumar Grover².

In order to better understand the quantum search problem, let us first classically define the search problem and then look at its quantum counterpart, comparing both using the *query complexity model*.

7.2 Query Model of Computation

Imagine your friend Alice has a word from a dictionary in her mind. Your task is to find this word by asking Alice only yes or no questions. For example, you can ask her, "Is the word 'exasperation'?" or "Is it between 'exaggerate' and 'diligent'?" etc. What is the least number of questions you should ask to find the word?

¹Even with the advantage of randomness in classical computing, we can show that we need at least $\Omega(n)$ queries by using [Yao's minimax principle](#).

²An Indian-American computer scientist who did his undergraduate studies at the Indian Institute of Technology (IIT), Delhi

In the above scenario, you are trying to *search* in a *structured data*. It is structured, as there is an underlying lexicographic ordering. Also, Alice represents what is called an *oracle*³. We know that for such ordered data, we can search classically in $\mathcal{O}(\log n)$ time, where n is the number of words in the dictionary, but this is not true for unstructured data. In the upcoming section, we will formalise the notion of an oracle and look into unstructured search.

Remarks. *Note that here structured data means that you know some predefined information about the data. For example, if the data is in ascending order, then you can exploit this information to search faster. Whereas when we say the data is unstructured, this means you know no predefined information about the data, which you can exploit to search faster.*

7.2.1 Classical Oracle

Consider a data set with N elements, for convenience, let $N = 2^n$. One can imagine the data stored as a list with consecutive elements in a contiguous memory location (this is generally how any data is stored in a computer). So, associated with each element, there is an index. One of the elements in this data is "marked", and we are interested in searching for this element. (For now, let us focus on searching for a single element. There are variants of Grover's algorithm where multiple elements can be marked as well. But the essential idea remains the same in both cases.)

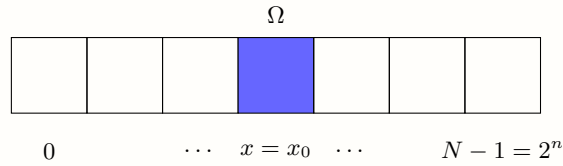


Figure 7.1: Unordered data with N elements

Suppose x_0 is the index of the "marked" element. Note that as $N = 2^n$, the indices can be n -bit long. Also, notice that there is a one-to-one correspondence between the elements and the indices. Thus, finding the "marked" element is equivalent to finding x_0 . Hereby, for the discussion, let us consider unstructured data.

As the data is unstructured, we know no predefined information. Thus, the classical algorithm to find x_0 is to simply go over the data set and check each element one by one. In terms of the above example, where you were questioning Alice, this time she has a word from a list of words from many different languages, where the words are jumbled up and not in any predefined order. Every time you can just ask her, "Is the word in this index?" or "that index," etc. Only she has access to the list and say yes or no to your questions.

³The word has origins from ancient Greece, where it means advice or information from the gods and often had a hidden meaning.

Formally, this can be captured by the following function:

$$f(x) = \begin{cases} 1, & \text{if } x = x_0 . \\ 0, & \text{otherwise.} \end{cases}$$

Here $f : \{0, 1\}^n \rightarrow \{0, 1\}$, a function that maps the n -bit strings indices to 1 or 0. Such a function is known as *black box function*.

Note that you do not have explicit knowledge of the function f . As if you did, you would already know x_0 . Instead, you only have access to a black box or oracle (for instance, Alice in the above example) that evaluates f on inputs x of your choice. Your goal is to find x_0 with the least number of queries to the black box.

Classically, we have to make, on average, $\mathcal{O}(N)$ queries to f in order to find x_0 .

Remarks. *Note that here we are considering the query model of computation. In this model, only the number of queries matters and not any other computational cost. Note that query complexity and time complexity are not equivalent, but query complexity gives a lower bound on the time complexity.*

Also, one must note that separations in the query complexity model do not directly imply separations in the time complexity model.

7.2.2 Quantum Oracle

How should one begin to think about search problems in a quantum computing setting? In particular, how are the elements represented, and what are the analogous indices here? What does 'searching' mean in this context?

In quantum mechanics, everything happens on a Hilbert space. A natural setup is to encode the indices of the elements as an orthonormal basis of the Hilbert space. So, given $N = 2^n$ elements, let us consider a Hilbert space of dimension $N = 2^n$, with each index encoded as one of the orthonormal bases. Without loss of generality, these indices can be encoded as $\{|0\rangle \dots |N-1\rangle\}$ (tensor product of n -qubits written in shorthand notation). One of the basis vectors is the index corresponding to the "marked" element, and our task is to find this. Note the similarity of this with the previously mentioned classical setup.

Initially, we have no clue about the index of the "marked" element. So, a natural starting point is to start with the state $|00\dots 0\rangle$ or an equal superposition of all the basis vectors, say $|\phi\rangle$.

$$|\phi\rangle = \frac{1}{\sqrt{2^n}} \sum_i |i\rangle$$

The task now is to devise unitary transformations that will transform the state $|\phi\rangle$ and take it *sufficiently* close to the index of the "marked" element (which is one of the orthogonal

basis states) so that when measured, we get this index with a *high* probability.

In the classical case, each time we made a guess, we had the classical oracle to say if it was correct or not. To have a quantum analogue of the classical oracle function f , we need to have a unitary that indicates the state we are looking for. Given $f(x)$, as seen in chapter 5, we can construct a unitary \mathbf{U}_f such that,

$$\mathbf{U}_f |x, b\rangle \rightarrow |x, b \oplus f(x)\rangle$$

Note that to query the i th position, we can set the input qubits to $|i, 0\rangle$.

Equivalently, if we set b as the $|+\rangle$ or $|-\rangle$ state, then the action of \mathbf{U}_f is simply to put a phase to the input state depending on the $f(x)$ value.

$$\begin{aligned}\mathbf{U}_f |x\rangle &= (-1)^{f(x)} |x, -\rangle \\ \mathbf{U}_f |x\rangle &= |x, +\rangle\end{aligned}$$

Thus, by applying the Hadamard gate to the target qubit b , we can change \mathbf{U}_f into an oracle that does the following:

$$\mathbf{U}'_f |x, b\rangle = (-1)^{f(x) \cdot b} |x, b\rangle$$

\mathbf{U}'_f is known as the *phase oracle*.

Thus, when $b = 1$, the quantum oracle will act on an n -qubit quantum state $|x\rangle$ and add a negative phase to the state if it is equal to the target state $|x_0\rangle$ and leaves it unchanged otherwise. Like how the classical oracle returned 1 when the state is $|x_0\rangle$, the action of adding negative phase can be thought of as an indication given by the quantum oracle. We will be using the oracle \mathbf{U}'_f in Grover's algorithm.

7.3 Grover's Search Algorithm

With the base set-up, our task is to find a unitary transformation that takes $|\phi\rangle = \frac{1}{\sqrt{2^n}} \sum_i |i\rangle$ to the index of the "marked" element, $|x_0\rangle$, given the quantum oracle \mathbf{U}'_f .

As $|x_0\rangle$ is also one of the basis states, if we measure $|\phi\rangle$ without doing anything, the probability that we get $|x_0\rangle$ is $\frac{1}{N}$, where $N = 2^n$. This probability is very small, and to increase this, we must transform $|\phi\rangle$ such that the coefficient of $|x_0\rangle$ increases thereby decreasing all other basis elements' coefficients.

Let $|x_0^\perp\rangle$ be an equal superposition of all the unmarked elements, which is orthogonal to $|x_0\rangle$,

$$|x_0^\perp\rangle = \frac{1}{\sqrt{N-1}} \sum_{i \neq x_0} |i\rangle$$

and we can write $|\phi\rangle$ as,

$$|\phi\rangle = c_1 |x_0\rangle + c_2 |x_0^\perp\rangle$$

and our goal is to increase c_1 while decreasing c_2 using suitable unitary transformation. So increasing c_1 corresponds to moving $|\phi\rangle$ closer to $|x_0\rangle$ on the plane spanned by $|x_0\rangle$ and $|x_0^\perp\rangle$, as shown in Fig. 7.2.

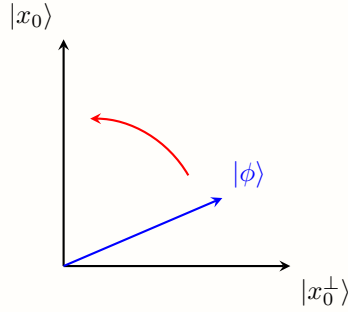


Figure 7.2: Goal is to move $|\phi\rangle$ closer to $|x_0\rangle$

Householder Transformation

- $2|a\rangle\langle a| - \mathbb{I}$ is rotation about $|a\rangle$:

One can write out the matrix corresponding to this expression and prove it is a reflection. Otherwise, intuitively, think about what it means to reflect a vector about the y -axis in an $x - y$ plane. It can be seen as adding a negative sign to the orthogonal x -component of that vector. Similarly, in any dimension, reflection about any state $|a\rangle$ is the same as adding a negative sign to all its orthogonal states. That is precisely what $2|a\rangle\langle a| - \mathbb{I}$ does.

- 2 reflections gives a rotation:

Imagine two lines, L_1 and L_2 in a 2D plane that intersect at an angle θ between them. For simplicity, let's align L_1 with the x -axis. This means the angle between L_1 and L_2 is simply the angle of L_2 , which we call θ . Now, take a vector \vec{v} . Let's say it makes an angle α with our first reflection line, L_1 . When we reflect \vec{v} across L_1 (the x -axis), its angle flips from α to $-\alpha$. Let's call this new vector \vec{v}' . Now, we reflect \vec{v}' (at angle $-\alpha$) across the second line, L_2 (at angle θ). A reflection flips a vector's angle relative to the reflection axis. The angle of \vec{v}' relative to L_2 is $(\theta - (-\alpha)) = \theta + \alpha$. To reflect it, we swing it to the other side of L_2 by that same amount. So, the final vector \vec{v}'' will be at an angle of $\theta + (\theta + \alpha) = 2\theta + \alpha$. Our original vector \vec{v} started at an angle α . The final vector \vec{v}'' is at an angle $2\theta + \alpha$. The total change is $(2\theta + \alpha) - \alpha = 2\theta$.

The state vector lies in the 2D plane spanned by the marked state $|x_0\rangle$ and the unmarked superposition $|x_0^\perp\rangle$. The two reflections are about the axis $|x_0^\perp\rangle$ (performed by the oracle) and the axis $|\psi\rangle$ (the initial state). If the angle

between these two reflection axes is α , then one Grover iteration rotates the state vector by an angle of 2α , moving it closer to the target state $|x_0\rangle$.

As the phase oracle adds a negative phase only to $|x_0\rangle$ component, its action is precisely $\mathbb{I} - 2|x_0\rangle\langle x_0|$, equivalently $2|x_0^\perp\rangle\langle x_0^\perp| - \mathbb{I}$. This is nothing but a reflection about the state $|x_0^\perp\rangle$. Note that this does not mean we know about $|x_0\rangle$, we simply have access to a black box that can do this for us.

Our goal is to rotate $|\phi\rangle$, and we have one reflection at hand, so all we need is another reflection. As two reflections result in a rotation, as explained in the above box. As we do not have any other information, one natural choice for another axis is to reflect on the equal superposition state $2|\psi\rangle\langle\psi| - \mathbb{I} = H^{\otimes n}(2|0\rangle\langle 0| - \mathbb{I})H^{\otimes n}$, where $|\psi\rangle = \frac{1}{\sqrt{2^n}}\sum_i |i\rangle$.

This rotation that we get by combining both the reflections is called a *Grover's iteration*,

$$G = H^{\otimes n}(2|0\rangle\langle 0| - \mathbb{I})H^{\otimes n}\mathbf{U}'_f$$

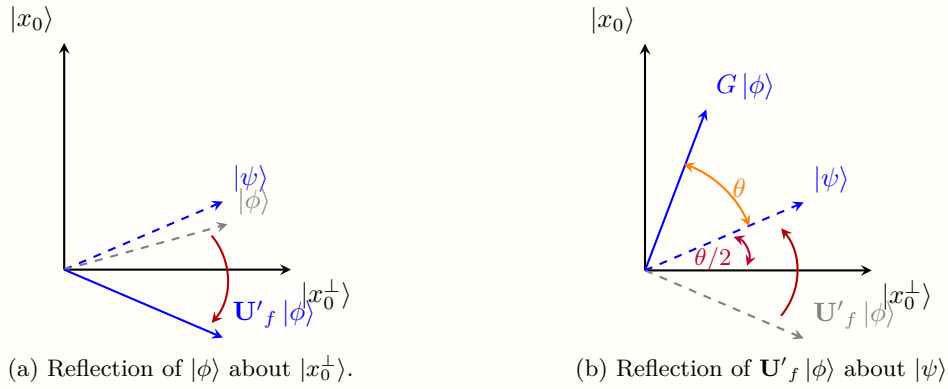


Figure 7.3: Overall rotation after first iteration of Grover's algorithm.⁴

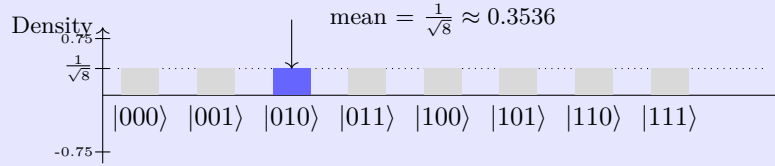
Now we just need to keep rotating the state a *sufficient number* of times and then measure it to get $|x_0\rangle$, the state we are searching for. Note that for each application of the Grover iteration G , we invoke the oracle \mathbf{U}'_f once. Thus, the number of times we rotate determines the query complexity of Grover's algorithm.

Remarks. Given we know we need to move $|\phi\rangle$ close to $|x_0\rangle$, why can't we move it in one go? We can not do this as recall we know nothing about f , thus we do not know x_0 . At each step, we are using the help of the oracle that puts a negative phase and slowly moves our vector towards x_0 . Another visual representation of this process is given in the box below.

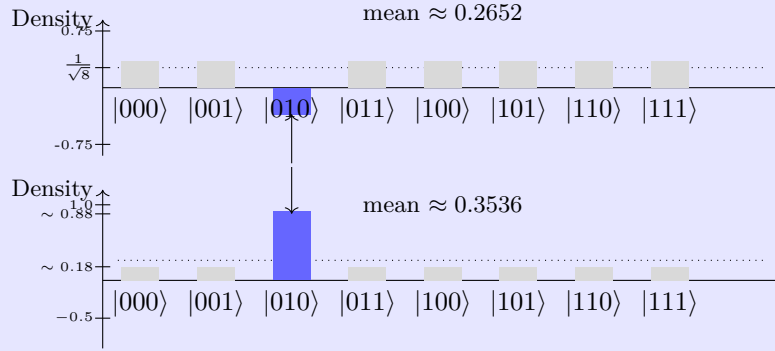
⁴In Fig. 7.3(a), $|\phi\rangle$ and $|\psi\rangle$ are exactly the same initially. For clarity, they are drawn slightly apart. Also, the angle θ is exaggerated for illustration. In reality, $|\phi\rangle$ starts very close to $|x_0^\perp\rangle$, so the rotation towards $|x_0\rangle$ at each iteration is generally small.

A simple application of Grover's Search Algorithm

For the sake of understanding, let us assume the coefficients of the basis states are real (in general, these are complex-valued). If we assume we have 8 elements and one is marked, then our initial $|\phi\rangle$ starts off with equal amplitude and has coefficients like those shown in the figure below.



The two operations in Grover's iteration are phase flip and reflection about the mean (i.e., $|\psi\rangle$). This corresponds to adding a negative sign to the marked amplitude and flipping about the dotted line shown in the figures below.



Notice how, after inverting the phase, flipping about the mean increases the amplitude of the marked state while decreasing the amplitude of all other states. This procedure is applied over and over again, increasing the amplitude of the marked state till it becomes more than $1/2$. As shown in the subsequent section, one can prove that this method takes \sqrt{N} steps to do this.^a

^aA more detailed explanation of this way of visualising Grover's algorithm can be found in the textbook *Dancing with Qubits* by Robert S. Sutor.

7.4 Query complexity of Grover's Search Algorithm

Suppose after k Grover iterations we want the state to be very close to $|x_0\rangle$, this implies the angle between the rotated state and $|x_0^\perp\rangle$ is very close to $\pi/2$. After one Grover iteration, from the Fig. 7.3, we see that the rotated state is at an angle $\theta + \theta/2$ from $|x_0^\perp\rangle$. Thus, after

k iterations we have,

$$\begin{aligned} k\theta + \frac{\theta}{2} &\approx \frac{\pi}{2} \\ \implies k &\approx \frac{\pi - \theta}{2\theta} \end{aligned}$$

As all the state vectors are unit vector $|\psi\rangle \cdot |x_0^\perp\rangle = \cos \theta/2$, this gives $\theta/2 = \cos^{-1} \sqrt{\frac{N-1}{N}}$. We are safe to assume the angle θ is small, therefore

$$\frac{1}{\sqrt{N}} = \sin \theta/2 \approx \theta/2$$

Substituting this we get $k = \mathcal{O}(\sqrt{n})$. Thus, in just $\mathcal{O}(\sqrt{n})$ queries, we can search the element, giving a quadratic speed-up compared to the classical computer.

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Chapter 8

Variational Quantum Algorithms

“If one proves the equality of two numbers a and b by showing first that a is less than or equal to b ; and then a is greater than or equal to b , it is unfair, one should instead show that they are really equal by disclosing the inner ground for their equality.”

– Emmy Noether, *Biography*

A common goal of variational algorithms is to find the quantum state with the lowest or highest eigenvalue of a certain observable. A key insight is taken from the variational theorem of quantum mechanics.

8.1 Variational Theorem

By the spectral theorem, a Hamiltonian being Hermitian can be written as,

$$\hat{\mathcal{H}} = \sum_{k=0}^{N-1} \lambda_k |\phi_k\rangle \langle \phi_k|$$

where N is the dimensionality of the space of states, λ_k is the k -th eigenvalue or, physically, the k -th energy level, and $|\phi_k\rangle$ is the corresponding eigenstate: $\hat{\mathcal{H}}|\phi_k\rangle = \lambda_k |\phi_k\rangle$, the expected energy of a system in the (normalized) state $|\psi\rangle$ will be:

$$\begin{aligned} \langle \psi | \hat{\mathcal{H}} | \psi \rangle &= \langle \psi | \left(\sum_{k=0}^{N-1} \lambda_k |\phi_k\rangle \langle \phi_k| \right) | \psi \rangle \\ &= \sum_{k=0}^{N-1} \lambda_k \langle \psi | \phi_k \rangle \langle \phi_k | \psi \rangle \\ &= \sum_{k=0}^{N-1} \lambda_k |\langle \psi | \phi_k \rangle|^2 \end{aligned}$$

If we take into account that $\lambda_0 \leq \lambda_k, \forall k$, we have:

$$\begin{aligned} \langle \psi | \hat{\mathcal{H}} | \psi \rangle &= \sum_{k=0}^{N-1} \lambda_k |\langle \psi | \phi_k \rangle|^2 \\ &\geq \sum_{k=0}^{N-1} \lambda_0 |\langle \psi | \phi_k \rangle|^2 \\ &= \lambda_0 \sum_{k=0}^{N-1} |\langle \psi | \phi_k \rangle|^2 \\ &= \lambda_0 \end{aligned}$$

Since $\{|\phi_k\rangle\}_{k=0}^{N-1}$ is an orthonormal basis, the probability of measuring $|\phi_k\rangle$ is $p_k = |\langle \psi | \phi_k \rangle|^2$, and the sum of all probabilities is such that $\sum_{k=0}^{N-1} |\langle \psi | \phi_k \rangle|^2 = \sum_{k=0}^{N-1} p_k = 1$. In short, the expected energy of any system is higher than the lowest energy or ground state energy:

$$\langle \psi | \hat{\mathcal{H}} | \psi \rangle \geq \lambda_0$$

The above argument applies to any valid (normalized) quantum state $|\psi\rangle$, so it is perfectly possible to consider parametrized states $|\psi(\vec{\theta})\rangle$ that depend on a parameter vector $\vec{\theta}$. This is where the "variational" part comes into play. If we consider a cost function given by $C(\vec{\theta}) := \langle \psi(\vec{\theta}) | \hat{\mathcal{H}} | \psi(\vec{\theta}) \rangle$ and want to minimize it, the minimum will always satisfy:

$$\min_{\vec{\theta}} C(\vec{\theta}) = \min_{\vec{\theta}} \langle \psi(\vec{\theta}) | \hat{\mathcal{H}} | \psi(\vec{\theta}) \rangle \geq \lambda_0$$

The minimum value of $C(\vec{\theta})$ will be the closest that one can get to λ_0 using the parametrized states $|\psi(\vec{\theta})\rangle$, and equality will only be reached if there exists a parameter vector $\vec{\theta}^*$ such that $|\psi(\vec{\theta}^*)\rangle = |\phi_0\rangle$.

If the (normalized) state $|\psi\rangle$ of a quantum system depends on a parameter vector $\vec{\theta}$, then the optimal approximation of the ground state (i.e. the eigenstate $|\phi_0\rangle$ with the minimum eigenvalue λ_0) is the one that minimizes the expectation value of the Hamiltonian $\hat{\mathcal{H}}$:

$$\langle \hat{\mathcal{H}} \rangle(\vec{\theta}) := \langle \psi(\vec{\theta}) | \hat{\mathcal{H}} | \psi(\vec{\theta}) \rangle \geq \lambda_0$$

The reason why the variational theorem is stated in terms of energy minima is that it includes a number of mathematical assumptions: - For physical reasons, a finite lower bound to the energy $E \geq \lambda_0 > -\infty$ needs to exist, even for $N \rightarrow \infty$. - Upper bounds do not generally exist.

However, mathematically speaking, there is nothing special about the Hamiltonian $\hat{\mathcal{H}}$ beyond these assumptions, so the theorem can be generalised to other quantum observables and their eigenstates provided they follow the same constraints. Also, note that if finite upper bounds exist, the same mathematical arguments could be made for maximising eigenvalues by swapping lower bounds for upper bounds. If the (normalized) state $|\psi\rangle$ of a quantum

system depends on a parameter vector $\vec{\theta}$, then the optimal approximation of the ground state (i.e. the eigenstate $|\phi_0\rangle$ with the minimum eigenvalue λ_0) is the one that minimizes the expectation value of the Hamiltonian $\hat{\mathcal{H}}$:

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Barren Plateau Problem: One major disadvantage of these variational quantum algorithms is the *barren plateau problem*. As long as there is some non-zero gradient, the parameters and the cost function keep changing, but what if there is a large gradient zero, which does not update the parameters or the cost function? There is a vast literature about this problem in classical machine learning as well as quantum computing¹.

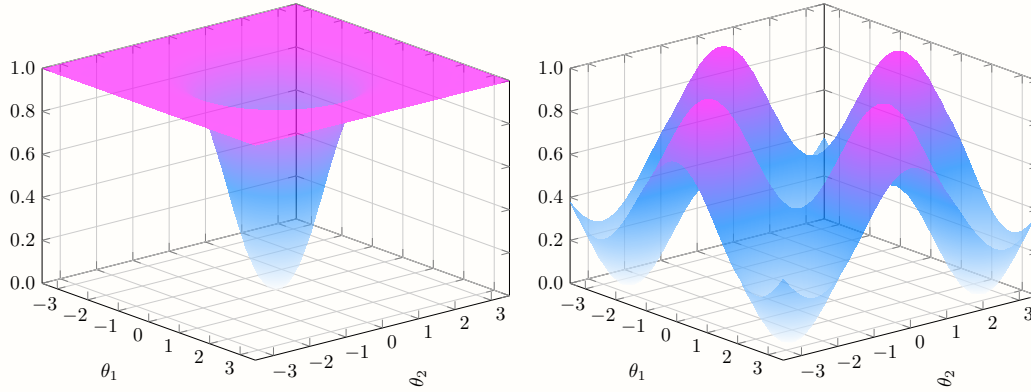


Figure 8.1: Two-dimensional cross-section through the landscape of cost functions.

8.2 Quantum Approximation Optimisation Algorithm

Quantum Approximate Optimisation Algorithm (QAOA) is a variational quantum algorithm designed for combinatorial optimisation problems.² Combinatorial optimisation is a

¹Refer to the paper [McClean et al. \[2018\]](#) or [Cerezo et al. \[2021a\]](#)

²First introduced in the paper [Farhi et al. \[2014\]](#)

field of mathematical optimisation, where one tries to find an optimal object from a finite set of objects.

The quantum circuit that implements these algorithms has its minima no less than that of the objective function that we are trying to optimise.

The overall scheme of QAOA can be roughly summarised by the following steps:

- **Map classical optimisation as a quantum problem:** The optimisation problem is encoded in the Hamiltonian of a quantum system, using some objective function C_z
- **Map the quantum problem to a parametrised quantum circuit:** A quantum circuit is constructed that encodes the potential solution
- **Classical sub-routine:** The parameters in the circuit are optimised classically to extremise the expectation value of the Hamiltonian
- **Measure:** The Final measured state provides solutions to the original optimisation problem
- **Repeat to improve accuracy:** This is repeated to improve the quality of the solution
- **Get the solution to the optimisation problem:** Final solution is represented in the computational basis, with the combinatorial solutions having the highest probabilities
- **Get good approximation ratio:** The goal is to find a solution such that $\frac{C_z}{C} \geq r$, where r is the approximation ratio, that is, the ratio of the solution given by our algorithm and the actual optimal solution for the problem.

In this chapter, we will see two graph optimisation theoretical problems- max cut and max independent set. To understand the problems better, let's first quickly go through some graph-theoretical definitions.

8.3 QAOA for Graph Theoretical Optimisation Problems

Graph $G = (V, E)$ is a set of vertices V and edges $E \subseteq V \times V$. A *cut* $(S, V \setminus S)$ is a partition of the vertex set into two disjoint subsets. The edges across the two parts of the cut, that is $e = (u, v) \in E$ such that $u \in S, v \in V \setminus S$ or vice versa, are called *cut edges*. A *maximum cut (max cut)* of a graph G is a cut that has the maximum number of cut edges. The MAX CUT problem is that given a graph $G = (V, E)$ we need to find the maximum cut of G .

A subset of vertices $U \subseteq V$ is called an *independent set* if no two vertices in U have an edge between them. Such a set U of maximum possible size for the given graph is called the *maximum independent set (max independent set)* of the graph. The MAX INDEPENDENT SET problem is given a graph G , to find the maximum independent set in the graph.

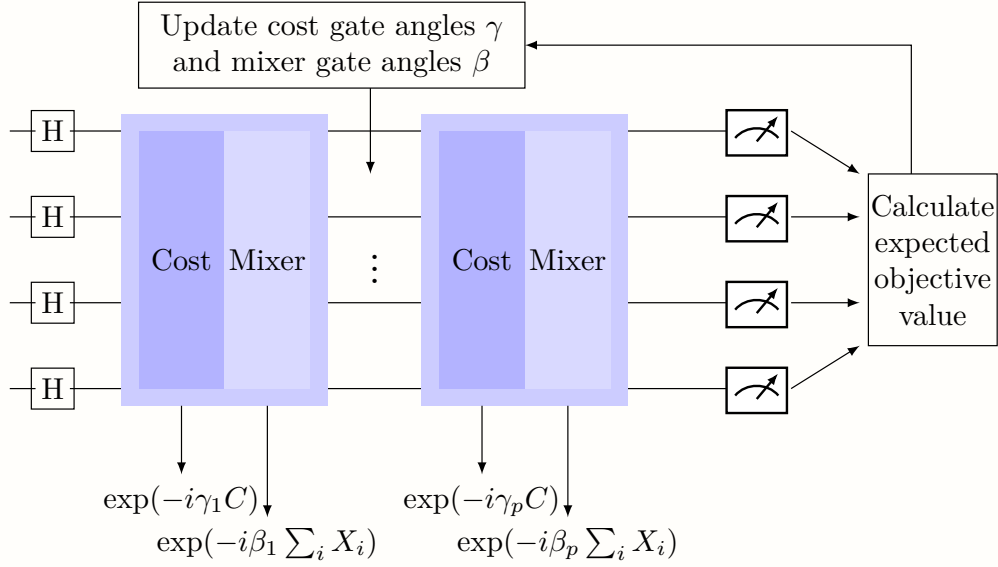


Figure 8.2: General working of QAOA

Both the MAX CUT and MAX INDEPENDENT SET problems are NP-HARD. One way of coping with this hardness is to devise approximation algorithms which give an approximate solution to the optimal one. For such approximation algorithms, we define something called the *approximation ratio*.

Definition 8.3.1 (Approximation Ratio). *The approximation ratio of an algorithm is the ratio of the value of solution returned by the algorithm to that of the optimum solution for the problem.*

In the subsequent section, we will see the best-known classical algorithms and the QAOA approach for these problems.

8.4 Max Cut

PROBLEM STATEMENT: Given a graph $G = (V, E)$, label the nodes as S or $V \setminus S$ such that the $(S, V \setminus S)$ is a max cut.

8.4.1 Classical Algorithm for Max Cut Problem

The best-known classical algorithm for this problem is the Goemans-Williamson algorithm. This is an approximation algorithm³ that produces a cut randomly, which, on expectation,

³Approximation algorithms are a class of algorithm that gives an approximate optimal solution rather than the exact solution. To know more, refer to the [Wikipedia page on approximation algorithm](#) or the textbook [Williamson and Shmoys \[2011\]](#)

is approximately 88% of the optimal max cut.

Given a graph $G = (V, E)$ with n vertices, let us denote these vertices as $V = \{1, 2 \dots n\}$. The max cut M for G , say $\text{maxcut}(G)$, can be seen as the solution of the following integer quadratic programming.

$$\begin{aligned} \text{maxcut}(G) = \max \frac{1}{2} \sum_{\{i,j\} \in E} (1 - x_i x_j) \\ \text{subjected to } x_i \in \{-1, 1\} \forall i \in V \end{aligned}$$

But solving the above quadratic programming is NP-Hard. So we try to "relax" the constraint in the hope of solving it. But how to "relax" the constraints?

Define $S^k \equiv \{x \in \mathbb{R}^{k+1} \mid \|x\| = 1\}$. Note that $S^0 = \{-1, 1\}$. With S^k defined, we can relax the above optimisation constraint from S^0 to S^k .

$$\begin{aligned} \text{maxcut}^{\text{SDP}}(G) = \max \frac{1}{2} \sum_{\{i,j\} \in E} (1 - x_i \cdot x_j) \\ \text{subjected to } x_i \in S^k \forall i \in V \end{aligned}$$

As we are now optimising over a bigger set S^k and not S^0 , $\text{maxcut}(G) \leq \text{maxcut}^{\text{SDP}}(G)$. These x_i s are now vectors in \mathbb{R}^n .⁴ Since $\|x_i\| = 1 \forall i \in V$ note that,

$$\begin{aligned} \frac{1}{4} \sum_{\{i,j\} \in E} \|x_i - x_j\|^2 &= \frac{1}{4} \sum_{\{i,j\} \in E} (x_i - x_j)(x_i - x_j) \\ &= \frac{1}{4} \sum_{\{i,j\} \in E} x_i \cdot x_i - 2x_i \cdot x_j + x_j \cdot x_j \\ &= \frac{1}{2} \sum_{\{i,j\} \in E} 1 - 2x_i \cdot x_j \end{aligned}$$

Remarks. Thus, solving for $\text{maxcut}^{\text{SDP}}(G)$ is equivalent to embedding the vertices on the sphere S^{n-1} so that the sum of the edge lengths is maximum.

8.4.2 Max Cut Semidefinite Programming

As we can efficiently solve semidefinite programming, we translate the problem to SDP (refer to Sec. 1.7 if unfamiliar with SDPs).

Given any graph G , we can construct its *adjacency matrix* A such that A_{ij} is 1 if there exists an edge between vertices i and j , and 0 otherwise. Also define a matrix X such that $X_{ij} = x_i \cdot x_j$. One can prove that this X is a positive semidefinite matrix as it can be written as $B^T B$ where B is a matrix whose columns are the vectors x_i .

⁴This way of translating vertices of a graph to \mathbb{R}^n or any other metric space is called metric embedding of the graph. Refer to [Matoušek \[2013\]](#) know more.

Theorem 8.4.1. $\{x_i\}_{i \in V}$ is optimal solution for $\text{maxcut}^{\text{SDP}}(G)$ if and only if X is optimal solution for the following SDP,

$$\begin{aligned} \min X \cdot A \\ \text{s.t. } X \succeq 0 \text{ and } X_{ii} = 1 \forall i \in V \end{aligned}$$

Proof. X feasible solution to the above SDP $\iff X_{ii} = 1 \forall i \in V \iff \|x_i\| = 1 \forall i \in V$, $\iff \{x_i\}_{i \in V}$ is optimal solution for $\text{maxcut}^{\text{SDP}}(G)$. Also as $\frac{1}{2} \sum_{\{i,j\} \in E} (1 - x_i \cdot x_j) = \frac{1}{2}|E| - \frac{1}{4}A \cdot X$. This shows minimising $X \cdot A$ corresponds to maximising the objective function of $\text{maxcut}^{\text{SDP}}(G)$. ■

Thus, by solving the above SDP (which can be solved efficiently on a classical computer), we can find an approximate solution to the MAX CUT problem. How "good" is this approximate solution?

Theorem 8.4.2. If H is a random hyperplane through origin and $\text{Cut}(H)$ is the size of the edge cut containing edges whose vertices are i and j such that x_i and x_j are separated by the hyperplane H , then $\exists \alpha \in \mathbb{R}$ with $\alpha \sim 0.868$ such that,

$$\mathbb{E}[\text{Cut}(H)] \geq \alpha \text{maxcut}^{\text{SDP}}(G) \geq \text{maxcut}(G)$$

Proof.

$$\mathbb{E}[\text{Cut}(H)] = \frac{\sum_{\{i,j\} \in E} \arccos(x_i \cdot x_j)}{\pi}$$

Let β be a constant such that $\arccos(t) \geq \beta(1-t) \forall t \in [-\pi/2, \pi/2]$. Solving for β and setting α as $2\beta/\pi$ gives $\alpha \sim 0.868$. Thus,

$$\begin{aligned} \mathbb{E}[\text{Cut}(H)] &= \frac{\sum_{\{i,j\} \in E} \arccos(x_i \cdot x_j)}{\pi} \\ &\geq \sum_{\{i,j\} \in E} \frac{\beta}{\pi} (1 - x_i \cdot x_j) = \alpha \text{maxcut}^{\text{SDP}}(G) \end{aligned}$$

■

8.4.3 Quantum Algorithm for Max Cut Problem

Let's first map this classical problem to a quantum problem. Let $x_i \in \{-1, 1\}$ be the label on the i^{th} node. The quantity $1 - x_i x_j$ is 0 if i, j have the same label and 2 if they have opposite.

$$H = -\frac{1}{2} \sum_{(i,j) \in E(G)} 1 - x_i x_j$$

Note that the negative sign in the expression converts the maximisation problem to a minimisation problem. We are formalising the problem in this way, as a minimisation problem, which can be seen as an energy minimisation problem, where the above H is the Hamiltonian (refer to chapter 2 for the definition of Hamiltonian). Thus, from a classical problem, we have translated it to a quantum problem of finding the lowest energy state or ground

state.

We have formulated the problem in a quantum mechanics language, but still this is not in a form a quantum computer can understand. The next step is to translate this Hamiltonian to the Hamiltonian of a quantum circuit.

The complete step-by-step process to translate H to a quantum circuit is beyond the scope of this book. Here, we only give an overarching idea of how this is done. This is done by translating the cost function to something called a *Quadratic Unconstrained Binary Optimisation (QUBO)*, which then gives a *cost function Hamiltonian* H_C for the circuit. This cost function Hamiltonian has the property that its minimum corresponds to the minimum of H and thus to the maximum cut value.

Once we have this H_C , our task is to prepare the ground state of H_C on the quantum computer. This is done by having a parametrised quantum circuit, whose parameters are updated as the circuit is run. Sampling from this state, with a high probability, yields the solution of our optimisation function. Initially, we start with a "guess state" for the ground state, which, with subsequent iteration and updation of parameters, finally converges close to the ground state. Most of the time, these parameters are optimised classically.

It turns out that the QUBO Hamiltonian is very similar to the Icing Hamiltonian that physicists are familiar with. This enables us to solve these optimisation problems by borrowing ideas from the Icing Hamiltonian.

Though for a general graph class, the QAOA algorithm for max cut is not known to exceed the performance of the classical Goemans-Williamson algorithm, it is still interesting to study due to the following approximation ratio guarantees, as shown in table 8.1.

Graph Class		Depth p	Approx. Ratio	Ref.
3-regular (worst-case)	graphs	1	≥ 0.6924	Farhi et al. [2014]
3-regular (worst-case, no cycles ≤ 5)	graphs with	2	≥ 0.7559	Wurtz and Love [2021]
3-regular (worst-case, no cycles $\leq 2p + 1$)	graphs with	3	≥ 0.7924 (conjectured)	Wurtz and Love [2021]

Table 8.1: Summary of some known approximation ratios of QAOA for Max-Cut on various graph classes and depths p .

8.5 Max Independent Set (MIS)

PROBLEM STATEMENT: Given a graph $G = (V, E)$, find a subset $U \subseteq V$ which is the maximum independent set.

8.5.1 Classical Algorithm for Max Independent Set Problem

Classically not just MIS is a hard problem, even approximating it is hard! For any constant $\epsilon > 0$ it is NP-Hard (unless $NP = ZPP$) to approximate MIS to within $n^{1-\epsilon}$ [Håstad \[1999\]](#); [Feige and Kilian \[1998\]](#) (This means the $\frac{\text{OPT(MIS)}}{\text{MIS given by algo}}$ is as large as $n^{1-\epsilon}$).

The best known classical approximation ratio is $O(n/(\log n)^2)$ [Boppana and Halldórsson \[1992\]](#), a bound that saw no improvement since 1992.

8.5.2 Quantum Algorithm for Max Independent Set Problem

Again, a full-length discussion about the QAOA quantum circuit for the max independent set problem is out of the scope of this book. Here, we will see how we translate the classical optimisation to a quantum problem.

Define a variable n_i for each node. It is 1 if the node is in the set A , and otherwise.

$$H = \sum_{(i,j) \in E(G)} n_i n_j$$

It is 0 if A is an independent set. There are multiple possible independent sets, and all get a value of 0 here.

Now, we add a term to split this degeneracy and separate out the maximum independent set.

$$H = \sum_{(i,j) \in E(G)} n_i n_j - \Delta \sum_{k \in V(G)} n_k$$

Here again, minimising H corresponds to the maximum independent set. Similar to the max cut problem, we can use Quadratic Unconstrained Binary Optimisation (QUBO) and construct a cost function Hamiltonian H_C for setting up the quantum circuit. The best known approximation ratios are given in table 8.2. Like the classical case, a similar hardness result is known that for a general graph, no polynomial time constant approximation algorithm is possible even with QAOA [Håstad \[1999\]](#); [Gamarnik \[2022\]](#).

Algorithm	Graph Type	Approx. Ratio	Ref.
QAOA $p = 1$	General (max degree Δ)	$1/(\Delta+1)$ (same as greedy)	Brady et al. [2023]
QAOA $p > 1$	General (random graphs)	Empirical ~ 0.66 – 0.90	Brady et al. [2023]
Iterative QAOA hybrids	General	Near-optimal empirically	Brady et al. [2023]
Rydberg Variational (QAA)	Unit-disk (geometric)	High quality; superlinear speedup	Ebadi et al. [2022]

Table 8.2: Summary of some known approximation ratios of QAOA for MIS on various graph classes and depths p .

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Part III

Quantum Information

Chapter 9

Generalising Operations

“If someone gave me a practical quantum computer tomorrow, then I confess that I can’t think of anything that I, personally, would want to use it for: only things that other people could use it for!”

– Scott Aaronson, *Quantum Computing since Democritus*

The dynamics of a closed quantum system are always assumed to be unitary. In other words, if initially the state of a quantum system is $|\psi\rangle$, then in a closed system this can evolve to $\mathbf{U}|\psi\rangle$ for some unitary matrix \mathbf{U} . It is known that quantum mechanics never creates or destroys information, but this is only true for closed systems. However, the assumption that a system is closed strays far from reality.

9.1 Preliminaries

We will discuss methodologies for understanding real-world systems, which can be regarded as an open system. By incorporating a system environment, we can view it as a closed system. We shall build the framework to study the evolution of a system, initially non-correlated with the environment, but that is entangled further on. The tools we have at our disposal are unitary evolution of the entire system, addition of a system through entanglement, and discarding a subsystem through a partial trace. We shall map the initially independent density operator of the system to the density operator of the system + environment as it interacts. The *superoperator* is introduced to transform from density operators of the system to density operators of the system + environment at a further time.

One slight caveat that should be mentioned is that if we have successive interactions from the environment, our framework only allows for independent interactions each time. To be clear, the initial state of the system must be completely uncorrelated from the environment, as we shall illustrate further. Only when we have multiple independent interactions, at each successive iteration, we can apply the superoperator formalism to note the transformation of the density matrix.

We begin with some emphasis on crucial topics that we shall work with for generalising operator transformations.

9.1.1 Schmidt Decomposition

Schmidt decomposition is essentially the Singular Value Decomposition (SVD) as mentioned earlier, applied to a quantum state that is shared between two separate systems, only for a bipartite state. Schmidt decomposition cannot be extended to systems composed of more than two parts. Its real power lies in how it elegantly reveals the fundamental connections and the entanglement between the two parts of the system.

The Schmidt decomposition theorem for a pure state $|\psi\rangle \in \mathcal{H} = \mathcal{H}_A \otimes \mathcal{H}_B$ of a bipartite quantum system, then there would exist orthonormal states $\{|l_A\rangle\}$ for \mathcal{H}_A , and $\{|l_B\rangle\}$ for \mathcal{H}_B , such that

$$|\psi\rangle = \sum_{l=1}^{\mathcal{R}} \sqrt{\lambda_l} |l_A\rangle |l_B\rangle$$

with λ_i positive real numbers satisfying $\sum_{i=1}^{\mathcal{R}} \lambda_i = 1$, where \mathcal{R} denotes the number of non-zero eigenvalues of the reduced density matrices $\rho_A = \text{Tr}_B \rho$ and $\rho_B = \text{Tr}_A \rho$. The states $|l_A\rangle$, and $|l_B\rangle$ depend on the particular state $|\psi\rangle$.

The Schmidt rank of the state $|\psi\rangle$ is the number of non-zero eigenvalues of the reduced density matrices, which are equal for ρ_A and ρ_B . Naively, the Schmidt number can be evidence for entanglement, but not a measure of entanglement, since a bipartite pure state is entangled if and only if its Schmidt number is greater than one. For a separable state, we have the Schmidt number equal to one.

9.1.2 Purification

On a related note, quantum state purification refers to the process of representing a mixed state as a pure quantum state of a higher-dimensional Hilbert space. The purification allows the original mixed state to be recovered by taking the partial trace over the additional degrees of freedom. The purification is not unique; there are different purifications that can lead to the same mixed states.

Given a quantum system described by a density matrix ρ_A , we can introduce another system ρ_B such that the state of the composite system is a pure state and $\rho_A = \text{Tr}_B \rho = \text{Tr}_B \{|\psi\rangle\langle\psi|\}$. To see this, consider a generic pure state for the global system given by the expression

$$|\psi\rangle = \sum_{lk} c_{lk} |l_A\rangle |k_B\rangle$$

with $\{|l_A\rangle\}$ and $\{|k_B\rangle\}$ as basis sets for the subsystems. The corresponding density matrix is thereby,

$$\rho = \sum_{lk} \sum_{l'k'} c_{lk} c_{l'k'}^* |l_A\rangle |k_B\rangle \langle l'_A| \langle k'_B|$$

whose trace can be evaluated as

$$\begin{aligned}
\rho_A &= \text{Tr}_B \rho = \sum_{k''} \langle k_B'' | \left(\sum_{lk} \sum_{l'k'} c_{lk} c_{l'k'}^* |l_A\rangle \langle k_B| \langle l'_A| \langle k'_B| \right) |k_B''\rangle \\
&= \sum_{k''} \sum_{lk} \sum_{l'k'} c_{lk} c_{l'k'}^* |l_A\rangle \langle k_B''| \langle k_B| \langle l'_A| \langle k'_B| k_B''\rangle \\
&= \sum_{lk} \sum_{l'k'} c_{lk} c_{l'k'}^* |l_A\rangle \langle l'_A| \underbrace{\langle k'_B| \left(\sum_{k''} |k_B''\rangle \langle k_B''| \right) |k_B\rangle}_{=\langle k'_B|k_B\rangle=\delta_{kk'}} \\
&= \sum_k \sum_{l'} c_{lk} c_{l'k}^* |l_A\rangle \langle l'_A|
\end{aligned}$$

Thus, the coefficients of the density matrix in the expansion in its subsystem must obey the relation

$$(\rho_A)_{ll'} = \sum_k c_{lk} c_{l'k}^*$$

which attributes correctly and has solutions, provided the Hilbert space of the adjoint system is large enough, with at least the same dimension. The existence of the solution is related to the size of the adjoint space, which is related to the entropy of the system. By increasing the size of the adjoint space, we always have a solution for the coefficients.

Purifying a qubit

Consider a qubit with density matrix ρ_A , and we adjoin another ancillary qubit for its purification. From the above condition, we have the following set of equations:

$$\begin{aligned}
(\rho_A)_{00} &= c_{00} c_{00}^* + c_{01} c_{01}^*, \\
(\rho_A)_{01} &= c_{00} c_{10}^* + c_{01} c_{11}^* = (\rho_A)_{10}^* \\
(\rho_A)_{11} &= c_{10} c_{10}^* + c_{11} c_{11}^*
\end{aligned}$$

which can be solved to give,

$$c_{00} = \sqrt{(\rho_A)_{00}}, \quad c_{01} = 0, \quad c_{10} = \frac{(\rho_A)_{01}^*}{\sqrt{(\rho_A)_{00}}}, \quad c_{11} = \sqrt{\frac{(\rho_A)_{10}(\rho_A)_{11} - |(\rho_A)_{01}|^2}{(\rho_A)_{00}}}$$

leading us to the purification. For a two-qubit system, it is thus possible to generate any density matrix ρ_A for one of the two qubits through unitary operations on that system.

9.2 Kraus Representation

Before studying the joint unitary evolution of completely decoupled systems, we consider a simplistic example that we have encountered earlier, being a closed system unitary evolution.

A closed system $|\psi\rangle$ evolves into $|\psi\rangle \rightarrow |\psi'\rangle = \mathbf{U}|\psi\rangle$ for some unitary matrix \mathbf{U} , through unitary evolution. The density matrix of the system is thus transformed from $\rho = |\psi\rangle\langle\psi|$ to $\rho' = (\mathbf{U}|\psi\rangle)(\langle\psi|\mathbf{U}^\dagger) = \mathbf{U}\rho\mathbf{U}^\dagger$. Generalizing to mixed states that evolve under unitary transforms as $|\varphi\rangle \mapsto |\varphi'\rangle = \mathbf{U}|\varphi\rangle$. Equivalently, we have the density matrix of the new state as

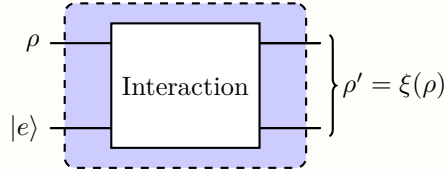
$$\begin{aligned}\rho' &= \sum_i p_i |\varphi'_i\rangle\langle\varphi'_i| \\ &= \sum_i p_i \mathbf{U}|\varphi_i\rangle\langle\varphi_i|\mathbf{U}^\dagger \\ &= \mathbf{U} \left(\sum_i p_i |\varphi_i\rangle\langle\varphi_i| \right) \mathbf{U}^\dagger \quad (\text{by linearity}) \\ &= \mathbf{U}\rho\mathbf{U}^\dagger\end{aligned}$$

As we see above, density matrices transform through the relation,

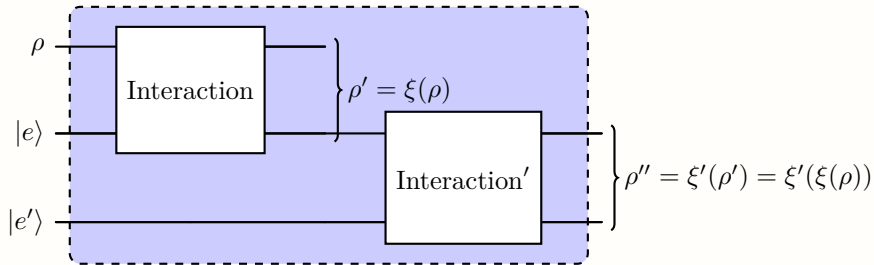
$$\rho \mapsto \rho' = \xi(\rho)$$

The *quantum operator* formalism generalises the dynamic change to a state that occurs as the result of some physical process, with ρ being the initial state before the process, and $\xi(\rho)$ the final state after the process occurs, possibly up to some normalisation factor.

A natural way to describe the dynamics of an open quantum system is to regard it as arising from an interaction between the system of interest, which we shall call the principal system, and an environment, which together form a closed quantum system.



For multiple interactions, we proceed iteratively, such that each interaction is independent, since we want the initial state of the system and the environment to be non-correlated.



We can generalise this notion of density operator mapping through the *Kraus operator-sum representation* as the map defined by a set of $\{E_k\}$ operators, with

$$\xi : \rho \rightarrow \rho' = \sum_k E_k \rho E_k^\dagger$$

such that the completeness relation is satisfied

$$\sum E_k^\dagger E_k = \mathbb{I}$$

which maps density operators to density operators obeying

- Hermiticity preserving:

$$\rho'^\dagger = \left(\sum_k E_k \rho E_k^\dagger \right)^\dagger = \sum_k (E_k^\dagger)^\dagger \rho^\dagger E_k^\dagger = \sum_k E_k \rho E_k^\dagger = \rho'$$

- Unit Trace preserving:

$$\text{Tr}(\rho') = \text{Tr} \left\{ \sum_k E_k \rho E_k^\dagger \right\} = \sum_k \text{Tr} \{ E_k \rho E_k^\dagger \} = \text{Tr} \left\{ \rho \sum_k E_k E_k^\dagger \right\} = \text{Tr} \rho = 1$$

- Positive semi-definiteness preserving:

$$\langle \phi | \rho' | \phi \rangle = \sum_k \langle \phi | E_k \rho E_k^\dagger | \phi \rangle \equiv \sum_k \underbrace{\langle \varphi_k |}_{\langle \varphi_k | = E_k \langle \phi |} \rho \underbrace{|\varphi_k \rangle}_{|\varphi_k \rangle = E_k^\dagger |\phi \rangle} \geq 0$$

Thereby, we have effectively mapped the joint evolution of a system and an adjoint environment as an entire closed system through the Kraus representation. Now, let us try building up the evolution of a generic state $|\psi_A\rangle$, by constructing an inner product preserving unitary operator U acting on a bigger system, through the superoperator as

$$U|\psi_A\rangle|0_B\rangle = \sum_k E_k|\psi_A\rangle|k_B\rangle$$

where $\{k_B\}$ represents an orthonormal basis for the extended system. Here, the unitarity naturally arises from the completeness relation of the Kraus operators.

For a physical understanding of the process of quantum operation, let us realise the equivalence of unitary operations on the global system. This structure gives rise to a probabilistic notion of a noisy channel, which we shall explore further later. Naively, we can define the probability of the k^{th} operator through $p(k) = \text{Tr}(E_k \rho E_k^\dagger)$, and the k^{th} density matrix can be normalized to give

$$\rho_k = \frac{E_k \rho E_k^\dagger}{\text{Tr}(E_k \rho E_k^\dagger)}$$

such that the quantum operation beautifully maps to a noisy communication channel where a state ρ is probabilistically replaced by the state ρ_k , with

$$\xi(\rho) = \sum_k E_k \rho E_k^\dagger = \sum_k \text{Tr}(E_k \rho E_k^\dagger) \frac{E_k \rho E_k^\dagger}{\text{Tr}(E_k \rho E_k^\dagger)} = \sum_k p(k) \rho_k$$

There exists an intricate structure to the notion of superoperators, where we can compose two Kraus operators ξ_A and ξ_B to give rise to

$$\xi = \xi_B \xi_A, \quad \xi(\rho) = \xi_B(\xi_A(\rho))$$

which gives rise to a *semigroup* structure due to the non existence of an invertible structure unless unitarity is maintained. The non-existence of invertibility gives rise to a physical notion of describing an evolution from t_0 to t_1 , but not the reverse. This can be seen as a loss of information from the system to the auxiliary adjoint system: environment, and we can't run the evolution backwards. This phenomenon gives rise to *decoherence* and will be delved into in detail further.

We also note that the different representations can give rise to the same superoperator. If two superoperators coincide $\xi(\rho) = \sum_k E_k \rho E_k^\dagger$ and $\xi'(\rho) = \sum_k F_k \rho F_k^\dagger$, if and only if there exists a unitary matrix \mathcal{U} such that

$$F_i = \sum_j \mathcal{U}_{ij} E_j$$

This can be shown by noting that two states produce the same density operator if there exists a unitary matrix transforming one state to the other, understood as an effective change of basis, but representing the same state, as

$$|\psi_i\rangle = \sum_j \mathcal{U}_{ij} |\varphi_j\rangle$$

This results in a unitary freedom in the operator sum representation.

Now, we are equipped to tackle the fundamental representation theorem:

Theorem 9.2.1. *A map $\xi : \rho \rightarrow \rho'$ satisfying the following requirements:*

- *linearity:* $\xi(p_A \rho_A + p_B \rho_B) = p_A \xi(\rho_A) + p_B \xi(\rho_B)$,
- *preserves hermiticity,*
- *preserves trace,*
- *is completely positive,*

has an operator-sum representation given by

$$\rho' = \sum_k E_k \rho E_k^\dagger$$

where E_k satisfy $\sum_k E_k^\dagger E_k = \mathbb{I}$,

For clarification, the *completely positive* is a stronger property than positive, which primarily constrains the positive nature of the density matrices. Complete positivity implies, for any extension of the Hilbert space \mathcal{H}_A , to $\mathcal{H}_A \otimes \mathcal{H}_B$, the superoperator $\xi \otimes \mathbb{I}$ must be positive.

Proof. Given a system \mathcal{H}_A satisfying the above axioms, we shall consider an auxiliary system \mathcal{H}_B of the same dimension. Let $|l_A\rangle$ and $|l_B\rangle$ be the orthonormal basis to define the maximally entangled state of $\mathcal{H}_A \otimes \mathcal{H}_B$ as

$$|\ell\rangle := \sum_l |l_A\rangle |l_B\rangle$$

Given a quantum operation ξ , we further define the operator on the maximally entangled state given by

$$\tilde{\xi} := (\mathbb{I}_A \otimes \xi)(|\ell\rangle\langle\ell|) = \sum_{ll'} (|l_A\rangle\langle l'_A|) \xi(|l_B\rangle\langle l'_B|)$$

The beauty of the argument relies on the notion that the operator $\tilde{\xi}$ provides a complete description of the quantum operation ξ . To understand the effect of ξ on an arbitrary state of \mathcal{H}_B , it is sufficient to know the action on the single maximally entangled state with the auxiliary system. To recover ξ from $\tilde{\xi}$, we note, for a state $|\psi_B\rangle = \sum_k c_k |k_B\rangle$ in \mathcal{H}_B , we define a corresponding state $|\psi_A\rangle$ in \mathcal{H}_A through

$$|\psi_A\rangle = \sum_k c_k^* |k_A\rangle$$

such that the effect of ξ can be recovered from $\tilde{\xi}$ through the partial trace as

$$\begin{aligned} \langle\psi_A|\tilde{\xi}|\psi_A\rangle &= \langle\psi_A|((\mathbb{I}_A \otimes \xi)(|\ell\rangle\langle\ell|))|\psi_A\rangle = \langle\psi_A|\left(\sum_{ll'} (|l_A\rangle\langle l'_A|) \xi(|l_B\rangle\langle l'_B|)\right)|\psi_A\rangle \\ &= \sum_k c_k \langle k_A| \left(\sum_{ll'} (|l_A\rangle\langle l'_A|) \xi(|l_B\rangle\langle l'_B|)\right) \sum_{k'} c_{k'}^* |k'_A\rangle \\ &= \sum_{ll'} \sum_{kk'} c_k c_{k'}^* \underbrace{\langle k_A|l_A\rangle}_{\delta_{lk}} \xi(|l_B\rangle\langle l'_B|) \underbrace{\langle l'_A|k'_A\rangle}_{\delta_{l'k'}} \\ &= \sum_{ll'} c_l c_{l'}^* \xi(|l_B\rangle\langle l'_B|) = \xi\left(\sum_l c_l |l_B\rangle \sum_{l'} c_{l'}^* \langle l'_B|\right) = \xi(|\psi_B\rangle\langle\psi_B|) \end{aligned}$$

Suppose now there exists some decomposition of $\tilde{\xi}$ as $\tilde{\xi} = \sum_i p_i |j_i\rangle\langle j_i|$ and we consequently define the map

$$E_i |\psi_B\rangle := \sqrt{p_i} \langle\psi_A|j_i\rangle$$

This linear map can be decomposed as

$$\begin{aligned} \sum_i E_i |\psi_B\rangle\langle\psi_B| E_i^\dagger &= \sum_i p_i \langle\psi_A|j_i\rangle\langle j_i|\psi_A\rangle \\ &= \langle\psi_A|\left(\sum_i p_i |j_i\rangle\langle j_i|\right)|\psi_A\rangle \end{aligned}$$

$$= \langle \psi_A | \tilde{\xi} | \psi_A \rangle = \xi(|\psi_B\rangle\langle\psi_B|)$$

Thereby, we have

$$\xi(|\psi_B\rangle\langle\psi_B|) = \sum_i E_i |\psi_B\rangle\langle\psi_B| E_i^\dagger$$

for all pure states $|\psi_B\rangle$ of \mathcal{H}_B . By convex linearity, it follows that $\xi(\rho) = \sum_i E_i \rho E_i^\dagger$. ■

In other words, the Kraus representation theorem infers that, if the evolution of a density matrix $\rho_B \rightarrow \rho'_B = \xi(\rho_B)$ preserves hermiticity and trace, is linear and completely positive, then the evolution can be realised by the unitary transformation, acting on a larger Hilbert space $\mathcal{H}_A \otimes \mathcal{H}_B$.

Representing Kraus Operators

Consider a single qubit quantum channel $|\psi\rangle = \alpha|0\rangle + \beta|1\rangle$ and an environment $|e\rangle = \sqrt{1-p}|0\rangle + \sqrt{p}|1\rangle$. The initial state of the system + environment (which evolves unitarily) is $|\Psi_0\rangle = |\psi\rangle \otimes |e\rangle$,

$$\begin{aligned} |\Psi_0\rangle &= (\alpha|0\rangle + \beta|1\rangle) \otimes (\sqrt{1-p}|0\rangle + \sqrt{p}|1\rangle) \\ &= \alpha\sqrt{1-p}|0,0\rangle + \alpha\sqrt{p}|0,1\rangle + \beta\sqrt{1-p}|1,0\rangle + \beta\sqrt{p}|1,1\rangle. \end{aligned}$$

Now, consider the interaction is through a CNOT gate with the control on the environment and the target on the system. We want to map out the evolution of the system with the environment through the interaction.

$$\left. \begin{array}{c} \rho \text{ --- } \text{---} \\ |e\rangle \text{ --- } \text{---} \end{array} \right\} \rho' = \xi(\rho)$$

Thus, the post-interaction joint state is

$$\begin{aligned} |\Psi\rangle &= \alpha\sqrt{1-p}|0,0\rangle + \alpha\sqrt{p}|1,1\rangle + \beta\sqrt{1-p}|1,0\rangle + \beta\sqrt{p}|0,1\rangle \\ &= \sqrt{1-p}(\alpha|0\rangle + \beta|1\rangle) \otimes |0\rangle + \sqrt{p}(\alpha|1\rangle + \beta|0\rangle) \otimes |1\rangle \\ &= \sqrt{1-p}|\psi\rangle \otimes |0\rangle + \sqrt{p}X|\psi\rangle \otimes |1\rangle. \end{aligned}$$

Tracing out the environment yields the reduced system state

$$\begin{aligned} \rho' &= \text{Tr}_e(|\Psi\rangle\langle\Psi|) \\ &= (1-p)|\psi\rangle\langle\psi| + pX|\psi\rangle\langle\psi|X \\ &= (1-p)\rho + pX\rho X. \end{aligned}$$

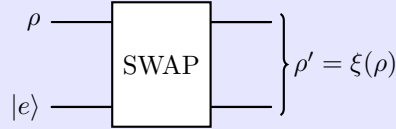
This is the familiar probabilistic bit-flip channel. In the Kraus (operator-sum) form, we may choose $E_0 = \sqrt{1-p}\mathbb{I}$ and $E_1 = \sqrt{p}X$, so that $\rho' = E_0\rho E_0^\dagger + E_1\rho E_1^\dagger$. We can check that $E_0^\dagger E_0 + E_1^\dagger E_1 = (1-p)\mathbb{I} + p\mathbb{I} = \mathbb{I}$.

The Kraus operator-sum representation $\rho \mapsto \xi(\rho) = \sum_k E_k \rho E_k^\dagger$ is derived under the assumption that the initial joint state factorises and is independent. If the system and environment are initially correlated, a reduced dynamical map that depends only on the system's density matrix need not exist. Thereby, the limitation of the formalism of a quantum operation is that systems that interact with degrees of freedom are used to prepare the system, even after the preparation is complete.

The Kraus description of reduced dynamics assumes an initial product state between system and environment, for a well-defined, completely positive map. When initial correlations are present, the reduced evolution can depend on which joint state was prepared. Consequently, the reduced map is not representable as a completely positive trace-preserving map acting only on the system's initial density operator.

Failure of Kraus representation

Consider a two-qubit system: the principal system S and an environment E . Let the global unitary be the SWAP operator (which interchanges the two qubits), through a combined unitary.



Now define two different *correlated* initial joint states $|\Psi_0\rangle_{(1)}$ and $|\Psi_0\rangle_{(2)}$ that have the same system marginal but different environment marginals.

$$|\Psi_0\rangle_{(1)} = \frac{1}{2}(|00\rangle\langle 00| + |11\rangle\langle 11|), \quad |\Psi_0\rangle_{(2)} = \frac{1}{2}(|00\rangle\langle 00| + |10\rangle\langle 10|).$$

Compute the system marginals (trace out E),

$$\text{Tr}_E(|\Psi_0\rangle_{(1)}) = \frac{1}{2}(|0\rangle\langle 0| + |1\rangle\langle 1|) = \frac{\mathbb{I}}{2}, \quad \text{Tr}_E(|\Psi_0\rangle_{(2)}) = \frac{1}{2}(|0\rangle\langle 0| + |1\rangle\langle 1|) = \frac{\mathbb{I}}{2}.$$

Thus, both joint states give the same system marginal state for the system, $\rho = \frac{\mathbb{I}}{2}$. Next compute the environment marginals (trace out S),

$$\text{Tr}_S(|\Psi_0\rangle_{(1)}) = \frac{1}{2}(|0\rangle\langle 0| + |1\rangle\langle 1|) = \frac{\mathbb{I}}{2}, \quad \text{Tr}_S(|\Psi_0\rangle_{(2)}) = |0\rangle\langle 0|.$$

Apply the same global unitary U_{SWAP} to both joint states and then trace out the environment to obtain the final system state. Because SWAP interchanges S and E , the reduced final state of S equals the original marginal of E ,

$$\rho'^{(i)} = \text{Tr}_E(U_{\text{SWAP}}|\Psi_0\rangle_{(i)}U_{\text{SWAP}}^\dagger) = \text{Tr}_S(|\Psi_0\rangle_{(i)})$$

Hence, we have,

$$\rho'^{(1)} = \frac{\mathbb{I}}{2}, \quad \rho'^{(2)} = |0\rangle\langle 0|$$

But the two initial joint states had the identical system marginal $\rho = \frac{\mathbb{I}}{2}$ and were subjected to the *same* global unitary. If a map ξ acting only on ρ (i.e. $\rho' = \xi(\rho)$) existed and were independent of initial correlations, it must produce a unique output for the input $\frac{\mathbb{I}}{2}$. The fact that the same input $\frac{\mathbb{I}}{2}$ leads to two different outputs $\frac{\mathbb{I}}{2}$ and $|0\rangle\langle 0|$ is a contradiction. Therefore, no dynamical map ξ depending only on ρ exists that reproduces the reduced dynamics for both correlated initial states. In particular, there is no Kraus representation $\rho' = \sum_k E_k \rho E_k^\dagger$ valid for these correlated preparations.

9.3 Generalised Measurements

A generalised measurement is described by a set $\{M_i\}$ of measurement operators, not necessarily self-adjoint, that satisfy the completeness relation

$$\sum_i M_i^\dagger M_i = \mathbb{I}$$

with the post-measurement state with outcome i is the

$$|\psi'_i\rangle = \frac{M_i|\psi\rangle}{\sqrt{\langle\psi|M_i^\dagger M_i|\psi\rangle}}$$

with probability of measurement

$$p_i = \langle\psi|M_i^\dagger M_i|\psi\rangle$$

We realise that the completeness relation results in unit probabilities. Further, the projective measurements described earlier are a special case of generalised measurements in which the operators M_i are orthogonal projectors with $M_i^\dagger = M_i$ and $M_i m_j = \delta_{ij} M_i$ with completeness $\sum_i M_i = \mathbb{I}$. Projective measurements together with unitary operations are equivalent to generalised measurements, in a bigger Hilbert space.

Positive Operator-Valued Measure (POVM)

Alice and Bob decide to play a quantum game. Alice has multiple copies of two qubits, each one either in the $|0\rangle$ state or the $|+\rangle$ state. She sends Bob one of these qubits and challenges him to find out which qubit was sent. What strategy do you think Bob can use to find out the qubit was sent correctly every single time? In the first place, is it possible for Bob to win this game every single time?

Suppose Bob decides to measure the qubit in the computational basis. Note that if the outcome is $|1\rangle$, then he can certainly say that $|0\rangle$ was not sent. But what if the outcome was $|0\rangle$? In this case, Bob will not be able to say for certain whether the qubit was $|0\rangle$ or $|+\rangle$. The same holds if he chooses to do the measurement in Pauli X eigenbasis (i.e. $\{|+\rangle, |-\rangle\}$ basis). Regardless of what set of orthogonal projective measurements Bob chooses, he will not be able to distinguish $|0\rangle$ and $|+\rangle$. Thus, non-orthogonal states can not be perfectly distinguished. So what can he best do? Is there a way to go beyond these orthonormal projective measurements? This brings us to the concept to POVM a broader class of measurements.

As mentioned in the last section of this chapter, generalized measurements need not require the condition $\Pi^2 = \Pi$, which is satisfied by the projective measurements. All we need is for the set of measurement operators to give well-defined probabilities.

Consider a set of positive operators $\mathcal{M} = \{E_1, E_2 \dots E_d\}$ such that $0 \leq E_i \leq \mathbb{I}$ for all $i \in \{1, 2 \dots, d\}$ and $\sum_i E_i = \mathbb{I}$. (Notice that the inequality is between matrices. $E_i \leq \mathbb{I} \implies \mathbb{I} - E_i \geq 0$, that is we require $\mathbb{I} - E_i$ to be a positive operator. The summation condition ensures that \mathcal{M} gives a valid probability distribution. Such a set of measurements is called *Positive Operator-Valued Measure (POVM)*.

Now with the power of POVM, if not distinguish every time, Bob can at least come up with a *zero error discrimination strategy*. Consider the following set of measurements:

$$\mathcal{M} = E_1, E_2, E_3$$

where $E_1 = \frac{\sqrt{2}}{1+\sqrt{2}}|1\rangle\langle 1|$, $E_2 = \frac{\sqrt{2}}{1+\sqrt{2}}|-\rangle\langle -|$ and $E_3 = \mathbb{I} - (E_1 + E_2)$, coefficients chosen such that the condition $0 \leq E_i \leq \mathbb{I}$ for all $i \in \{1, 2 \dots, d\}$.

Using this when Bob does the measurement, if the outcome is that of E_1 , he can certainly say Alice sent $|+\rangle$ state, and if it was E_2 again with certainty, he can say she sent $|0\rangle$ state. But if the outcome is that of E_3 , then he will not be able to say which state it was. Thus, whenever he is able to find out the state, he can do it with complete certainty. Such strategies are called *zero error discrimination strategy*, which is one of many advantages opened up by POVM measurements.

Further Reading & References

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Chapter 10

Quantum Entropy

“The theory ‘All crows are black’ is refuted by the single observation of a white crow, while the theory ‘Some crows are black’ is not refuted by the observation of a thousand white crows.”

– F. Bavaud, *Information theory (paraphrasing Popper)*

10.1 Shannon Entropy

Suppose a random variable X can take values $x_1, x_2 \dots x_n$ with some probabilities $p_1, p_2 \dots p_n$ respectively. How can one quantify how much information is gained by knowing the value of X ?

Intuitively, we would want the information function, say $I(X)$, to depend on p_i 's and not the labels x_i 's, as the event of 50% head and 50% tail occurrence should contain the same information as the event of 50% one and 50% zero. Also, the information content should not have drastic jumps or falls with slight tweaks in the probabilities. And it is also reasonable to expect that the information gained when two independent events occur with individual probabilities p and q is the sum of the information gained from each event alone.

One can show that the function $I(p) = k \log p$ for some constant k satisfies all the above-stated intuitive conditions. More formally $I(p)$ follows:

- I is a function of $p_1, p_2 \dots p_n$ and not $x_1, x_2 \dots x_n$,
- $I(p)$ is a smooth function,
- $I(pq) = I(p) + I(q)$.

Given the above intuition, let us see the definition of *Shannon entropy*. Also note that when we say $\log N$ we always mean logarithm to the base 2. This is adopted as in the most basic form, information in the current day digital computers is represented as 1 or 0, in other words as *bits*.

Definition 10.1.1. (Shannon entropy) Given a probability distribution p_1, p_2, \dots, p_n , the Shannon entropy associated with this probability distribution is

$$H(X) \equiv H(p_1, p_2, \dots, p_n) \equiv - \sum_i p_i \log p_i$$

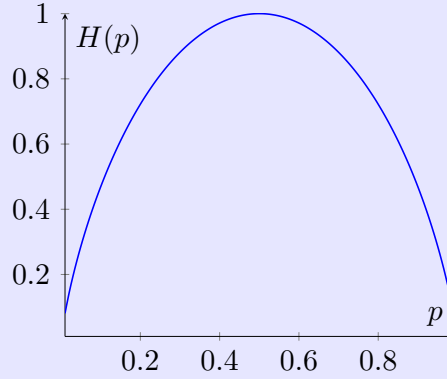
Events that never occur, $p_i = 0$, are not considered in the calculation of entropy as intuitively they do not add to the information content of E (More rigorously, one can also argue that $\lim_{p_i \rightarrow 0} p_i \log p_i = 0$). On average, Shannon entropy quantifies the information gain when we learn the value of a contextual bit in a message.

Shanon Binary entropy

Consider a two-state system for $n = 2$ and define $p_1 = p$ where $0 \leq p \leq 1$, hence $p_2 = 1 - p$, thereby, the Shanon binary entropy is a function of p alone as

$$H(p_1 = p, p_2 = 1 - p) = -p \log p - (1 - p) \log(1 - p)$$

which can be visualized graphically, in a simple plot



Note that the entropy equals zero at $p = 0$ or $p = 1$, and attains the maximal value when $p = \frac{1}{2}$. This is consistently well defined as a notion of entropy since it processes the average information content of each letter in a message. Information is a measure of *a priori ignorance*. If we already know that we shall receive a with certainty ($p = 1$), then no information is gained from its reception, and similarly for receiving b when $p = 0$. For the equiprobable case, ignorance is maximum; hence, the maximum possible information is available.

The above example can be elaborated to a general case to understand that $H(p_1, p_2, \dots, p_n)$ is maximum when $p_1 = p_2 = \dots = p_n = \frac{1}{n}$.

Defining a quantity called *relative entropy* is equally useful, which can measure how close two probability distributions are.

Definition 10.1.2. (*Relative entropy*) Given two probability distributions p and q , the relative intensity is defined as

$$H(p||q) \equiv \sum_i p_i \log \frac{p_i}{q_i} \equiv -H(X) - \sum_i p_i \log q_i$$

Conventionally, $\lim_{q_i \rightarrow 0} -p_i \log q_i \rightarrow \infty$. The motivation for defining relative entropy as above comes from the following theorem.

Theorem 10.1.1. *The relative entropy is non-negative and is zero if and only if the two probability distributions are equal.*

Proof. Using the identity that $\log x = \frac{\ln x}{\ln 2} \leq x - 1$ we can write

$$\begin{aligned} H(p||q) &= - \sum_i p_i \log \frac{q_i}{p_i} \geq \frac{1}{\ln 2} \sum_i p_i \left(1 - \frac{q_i}{p_i}\right) \\ &= \frac{1}{\ln 2} \sum_i (p_i - q_i) = \frac{1}{\ln 2} (1 - 1) = 0 \end{aligned}$$

We can see that equality is held when $p_i = q_i$ for all x_i 's. ■

Corollary 10.1.1.1. *If X has d outcomes, then $H(X) \leq \log d$ with equality if and only if X is a uniform distribution.*

10.2 Classical Data Compression

Can information be efficiently stored by compressing the bit string when given in a series of bit strings? If possible, to what fraction can we compress? As a first insight into classical data compression, we employ a lossless data compression technique that assigns variable-length codes to characters based on their frequency of occurrence in the data.

Huffman Data Encoding

Huffman coding uses a greedy algorithm to build a prefix tree that optimises the encoding scheme so that the most frequently used symbols have the shortest encoding. Consider a message written in the alphabet such that the frequency of occurrence of different letters is different, due to a probability distribution. To send a code word, we need $\sum_i p_i l_i$ bits where l_i is the length, in bits, of the coded letter. Note that the good strategy, here as in any other useful compression code, is to encode the most probable strings in the shortest sequences and the less probable strings in the longest sequences.

To address these questions mathematically rigorously, let us consider an information source that produces independent and identically distributed bits $X_1, X_2 \dots$ each of which is zero with probability p and one otherwise. Though sources often do not behave in the real world in this fashion, this is a good approximation and works well in most cases.

For a more concrete understanding, consider X_i as the i^{th} coin toss with a head occurring with $p = 0.4$. We know that in the large n limit, we will likely find 0.4 fraction of the tosses to be heads and the remaining tails. We call such sequences *typical sequences*. Formally defined as follows.

Definition 10.2.1. (*Typical sequence*) Given $X_1 \dots X_n$ with each X_i equal to 0 with probability p and 1 with probability $1 - p$. In the large n limit, we expect with high probability a fraction p of the X_i 's to be zero and the remaining ones. A sequence $x_1 \dots x_n$ for which this is true is called a *typical sequence*. Those that do not follow this are called *atypical sequences*.

Using the fact that the information source produces independent X_i that will highly likely be typical sequences with large n , we get

$$p(x_1, \dots, x_n) = p(x_1)p(x_2) \dots p(x_n) \approx p^{np}(1-p)^{(1-p)n}$$

How many bits do we need to represent this sequence? Taking logarithms on both sides, we find that

$$-\log p(x_1, \dots, x_n) \approx -np \log p - n(1-p) \log(1-p) \equiv nH(X)$$

Thus, $p(x_1, \dots, x_n) \approx 2^{-nH(X)}$ from which we can say that there are at most $2^{nH(X)}$ typical sequences (as total probability of all typical sequences ≤ 1). Therefore, we can use only $nH(X) \leq n$ bits to identify these typical sequences uniquely. In this sense, we can say that the information content is not n bits but $nH(X)$, and per bit it is $H(X)$.

Definition 10.2.2. (*Entropy rate*) Given a random variable X distributed according to the source distribution, $H(X) = -p \log p - (1-p) \log(1-p)$ is called the *entropy of the source distribution* or the *entropy rate of the source*.

In other words, for such an independent and identically distributed information source, in the large n limit, the data from n bits can be compressed to $nH(X)$ bits. One could make this idea more general by defining ϵ -typical strings.

Definition 10.2.3. Given a $\epsilon > 0$ we say a string is ϵ -typical if

$$2^{-n(H(X)+\epsilon)} \leq p(x_1, \dots, x_n) \leq 2^{-n(H(X)-\epsilon)}$$

$|T(n, \epsilon)|$ denotes the set of all ϵ -typical sequences.

Definition 10.2.4. (*Compression and decompression scheme*) A compression scheme, $C^n(x)$, of rate R maps the possible sequences of $x = x_1, \dots, x_n$ to $\lfloor nR \rfloor$ length bit strings. The corresponding decompression scheme, $D^n(x)$, takes the nR length string to n length string. A compression-decompression scheme is called *reliable* if the probability of $D^n(C^n(x)) = x$ approaches one as n tends to ∞ .

The following theorem shows that $H(X)$ is necessary and sufficient to store the output from the source reliably. Before that, we will see a useful lemma whose proof relies on the law of large numbers.

Lemma 10.2.1. *For a fixed $\epsilon > 0$ and any $\delta > 0$ with sufficiently large n , the probability that the sequence is ϵ -typical is at least $1 - \delta$. When we fix both $\epsilon > 0$ and $\delta > 0$ then*

$$(1 - \delta)2^{n(H(X) - \epsilon)} \leq |T(n, \epsilon)| \leq 2^{n(H(X) + \epsilon)}$$

If $S(n)$ is a collection at most 2^{nR} strings of length n , where $R < H(X)$ and n large, then for any $\delta > 0$

$$\sum_{x \in S(n)} p(x) \leq \delta$$

The probability of finding a string from this set goes to zero with large n .

Theorem 10.2.2. *(Shannon's noiseless channel coding theorem) Suppose $X = x_1, \dots, x_n$ is an independent and identically distributed information source, and $H(X)$ is the entropy rate, then a reliable compression-decompression scheme exists if and only if $R > H(X)$.*

Proof. As noted, a typical sequence is an n -letter message, $X = x_1, \dots, x_n$, where $x_i \in \mathcal{A}$, and we have an independent distribution of letters with specific probabilities, such that we have np_i times the i^{th} letter on average. The number of such strings can be enumerated as

$$\frac{n!}{\prod_{i=1}^k (np_i)!}$$

, which represents the number of distinct strings, having the requisite number of parameters. We can show that this number must approximate

$$\frac{n!}{\prod_{i=1}^k (np_i)!} \approx 2^{nH(p_1, \dots, p_k)}$$

explicitly shown using the Stirling's formula. Thus, the probability of obtaining such a typical sequence is the inverse.

Thereby, we obtain,

$$-\frac{1}{n} \log p(x_1, \dots, x_k) = -\frac{1}{n} \sum_{i=1}^n \log(p(x_i)) \approx H(p_1, \dots, p_k)$$

where the last (approximate) equality is guaranteed by the law of large numbers. The frequency $\frac{n_j}{n}$ of the letter j in the message is substituted by the a priori probability p_j , such that we obtain the number of times j appears in the message.

The law of large numbers also leads us to, for $\epsilon > 0$, we say a sequence is ϵ -typical, when

$$\left| -\frac{1}{n} \log p(x_1, \dots, x_n) - H(p_1, \dots, p_k) \right| < \epsilon$$

as defined earlier. Then, for any $\delta > 0$, the probability that a given sequence is ϵ -typical is larger than $1 - \delta$, for sufficiently large n . Therefore, most of the sequences are ϵ -typical in the limit of large n .

Since there are $2^{nH(X)}$ typical sequences, asymptotically in n , each occurring with a probability $2^{-nH(X)}$, we can identify which one of these sequences actually occurred using $nH(X)$ bits. Thus, asymptotic compression to $H(X)$ bits per letter is optimal. ■

10.3 Von Neumann Entropy

Like how Shannon's entropy measures the information content of classical probability distributions, *Von Neumann entropy* is defined for quantum states.

Definition 10.3.1. (*Von Neumann entropy*) Given a quantum state's density matrix ρ , its *Von Neumann entropy* is defined as

$$\mathcal{S}(\rho) := -\text{Tr}\{\rho \log \rho\}$$

The above definition is motivated by the fact that it resembles the classical Shannon's entropy when expressed in terms of the eigenvalues of ρ , say λ_i 's, as

$$\mathcal{S}(\rho) = -\text{Tr}\{\rho \log \rho\} = -\sum_i \lambda_i \log \lambda_i$$

To compare the entropy of two density matrices, similar to the notion of classical relative intensity, *quantum relative entropy* is defined.

Definition 10.3.2. (*Quantum relative density*) Given two density matrices ρ and σ the *quantum relative density* is defined as

$$\mathcal{S}(\rho||\sigma) := \text{Tr}\{\rho \log \rho\} - \text{Tr}\{\rho \log \sigma\}$$

Asymmetry in the relative entropy measure

The above defined relative entropy measures, both classical $H(p||q)$ and quantum $\mathcal{S}(\rho||\sigma)$, is asymmetric in p and q (ρ and σ). In some cases, the logarithm diverges. Thus, given two probability distributions (or density matrices), we choose one to be p (ρ) and the other to be q (σ) in such a way that the logarithm term makes sense. The asymmetry in relative entropy arises because it measures the difference in information between two probability distributions, not just the distance between them. In essence, this asymmetry is not a deficiency but a feature, arising from the inherent asymmetry in the mathematical models from which both concepts emerge.

Definition 10.3.3. (*Kernel and Support of density matrix*) The vector space spanned by the eigenvectors of the density matrix ρ with eigenvalue zero is called the *kernel*, and that spanned by the non-zero eigenvectors is called the *support*.

If the kernel of σ intersects the support of ρ non-trivially, then relative entropy is $+\infty$.

Theorem 10.3.1. *The quantum relative entropy is non-negative and is zero if and only if the two density matrices are equal.*

Proof. Let the spectral decomposition of the density matrix ρ be $\rho = \sum_i \lambda_i |u_i\rangle\langle u_i|$, where $|u_i\rangle$ are the orthonormal eigenvectors and λ_i are the corresponding non-negative eigenvalues summing to one ($\sum_i \lambda_i = 1$).

We can write the relative entropy as:

$$\mathcal{S}(\rho||\sigma) = \text{Tr}\{\rho(\log \rho - \log \sigma)\}$$

Let's evaluate the two terms separately in the eigenbasis of ρ :

$$\text{Tr}\{\rho \log \rho\} = \text{Tr} \left\{ \left(\sum_i \lambda_i |u_i\rangle\langle u_i| \right) \left(\sum_j (\log \lambda_j) |u_j\rangle\langle u_j| \right) \right\} = \sum_i \lambda_i \log \lambda_i$$

$$\text{Tr}\{\rho \log \sigma\} = \text{Tr} \left\{ \left(\sum_i \lambda_i |u_i\rangle\langle u_i| \right) \log \sigma \right\} = \sum_i \lambda_i \langle u_i | (\log \sigma) | u_i \rangle$$

Now, let the spectral decomposition of σ be $\sigma = \sum_j \mu_j |v_j\rangle\langle v_j|$. Then $\log \sigma = \sum_j (\log \mu_j) |v_j\rangle\langle v_j|$. Substituting this into the expression for $\text{Tr}\{\rho \log \sigma\}$:

$$\langle u_i | (\log \sigma) | u_i \rangle = \sum_j \langle u_i | (\log \mu_j) | v_j \rangle \langle v_j | u_i \rangle = \sum_j (\log \mu_j) |\langle u_i | v_j \rangle|^2$$

Let's define $\mathcal{P}_{ij} = |\langle u_i | v_j \rangle|^2$. Note that for any fixed i , $\sum_j \mathcal{P}_{ij} = \sum_j \langle u_i | v_j \rangle \langle v_j | u_i \rangle = \langle u_i | (\sum_j |v_j\rangle\langle v_j|) | u_i \rangle = \langle u_i | I | u_i \rangle = 1$.

The relative entropy is then:

$$\mathcal{S}(\rho||\sigma) = \sum_i \lambda_i \log \lambda_i - \sum_i \lambda_i \left(\sum_j \mathcal{P}_{ij} \log \mu_j \right)$$

The function $\log(x)$ is strictly concave. By Jensen's inequality, for each i :

$$\sum_j \mathcal{P}_{ij} \log \mu_j \leq \log \left(\sum_j \mathcal{P}_{ij} \mu_j \right)$$

Let's define a probability distribution $q_i = \sum_j \mathcal{P}_{ij} \mu_j = \langle u_i | \sigma | u_i \rangle$. The set $\{q_i\}$ forms a probability distribution since $\sum_i q_i = \sum_i \langle u_i | \sigma | u_i \rangle = \text{Tr}(\sigma) = 1$.

Substituting this back into the expression for relative entropy, we get a lower bound:

$$\mathcal{S}(\rho||\sigma) \geq \sum_i \lambda_i \log \lambda_i - \sum_i \lambda_i \log q_i = \sum_i \lambda_i \log \left(\frac{\lambda_i}{q_i} \right)$$

This final expression is exactly the classical relative entropy (or Kullback-Leibler divergence) $H(\lambda||q)$ between the probability distribution of eigenvalues of ρ , $\{\lambda_i\}$, and the distribution of the diagonal elements of σ in the eigenbasis of ρ , $\{q_i\}$. As established in the chapter, the classical relative entropy is non-negative, $H(\lambda||q) \geq 0$.

For the equality $\mathcal{S}(\rho||\sigma) = 0$ to hold, two conditions must be met

1. The inequality $H(\lambda||q) \geq 0$ must be an equality. This happens if and only if $\lambda_i = q_i$ for all i . So, $\lambda_i = \langle u_i|\sigma|u_i\rangle$.
2. The Jensen's inequality for the concave log function must be an equality for every i . This occurs if and only if for each i , all the values of μ_j for which $\mathcal{P}_{ij} = |\langle u_i|v_j\rangle|^2 > 0$ are identical.

The second condition implies that for any given eigenvector $|u_i\rangle$ of ρ , all eigenvectors $|v_j\rangle$ of σ that it has a non-zero projection on must share the same eigenvalue. This is only possible if each $|u_i\rangle$ is also an eigenvector of σ . Since $\{|u_i\rangle\}$ forms a basis, this means that ρ and σ must commute and are thus simultaneously diagonalizable.

If they share the same set of eigenvectors, let this basis be $\{|k\rangle\}$. Then $\rho = \sum_k \lambda_k |k\rangle\langle k|$ and $\sigma = \sum_k \mu_k |k\rangle\langle k|$. In this case, the quantum relative entropy simplifies to the classical relative entropy of their eigenvalues

$$\mathcal{S}(\rho||\sigma) = \sum_k \lambda_k \log \left(\frac{\lambda_k}{\mu_k} \right)$$

This is zero if and only if $\lambda_k = \mu_k$ for all k . Since they have the same eigenvalues and the same corresponding eigenvectors, the density matrices must be identical, $\rho = \sigma$. ■

Theorem 10.3.2. *The following are some properties of $\mathcal{S}(\rho)$:*

1. $\mathcal{S}(\rho)$ is non-negative. It is zero if and only if ρ is pure state.
2. In a d -dimensional Hilbert space the entropy is at most $\log d$. It is equal to $\log d$ if and only if the state is a completely mixed state.
3. Suppose a composite system $\mathcal{H} = \mathcal{H}_A \otimes \mathcal{H}_B$ is in pure state then $\mathcal{S}(A) = \mathcal{S}(B)$.
4. Suppose p_i are probabilities of the state being in ρ_i then

$$\mathcal{S} \left(\sum_i p_i \rho_i \right) = H(p_i) + \sum_i p_i \mathcal{S}(\rho_i)$$

Proof. 1. The von Neumann entropy is defined in terms of the eigenvalues $\{\lambda_i\}$ of the density matrix ρ as $\mathcal{S}(\rho) = -\sum_i \lambda_i \log \lambda_i$. For a density matrix, the eigenvalues satisfy $0 \leq \lambda_i \leq 1$. For any λ_i in this range, $\log \lambda_i \leq 0$. Thus, each term $-\lambda_i \log \lambda_i$ is non-negative. The sum of non-negative terms is also non-negative, so $\mathcal{S}(\rho) \geq 0$.

The equality $\mathcal{S}(\rho) = 0$ holds if and only if every term in the sum is zero. A term $-\lambda_i \log \lambda_i$ is zero if $\lambda_i = 0$ or $\lambda_i = 1$. Since the eigenvalues must sum to one ($\sum_i \lambda_i = 1$), it must be that exactly one eigenvalue is 1 and all others are 0. A density matrix with this eigenvalue distribution describes a pure state, $\rho = |\psi\rangle\langle\psi|$. Conversely, if ρ is a pure state, its eigenvalues are $\{1, 0, \dots, 0\}$, and its entropy is $\mathcal{S}(\rho) = -1 \log 1 - \sum 0 \log 0 = 0$.

2. We want to maximize $\mathcal{S}(\rho) = -\sum_{i=1}^d \lambda_i \log \lambda_i$ in a d -dimensional space. We can use the non-negativity of the relative entropy. Let ρ be any state and let $\sigma = \frac{1}{d}I$ be the completely mixed state. From Klein's inequality, $\mathcal{S}(\rho||\sigma) \geq 0$.

$$\text{Tr}\{\rho \log \rho\} - \text{Tr}\{\rho \log \sigma\} \geq 0$$

$$-\mathcal{S}(\rho) - \text{Tr}\left\{\rho \log \left(\frac{1}{d}I\right)\right\} \geq 0$$

$$-\mathcal{S}(\rho) - \text{Tr}\{\rho(\log(1/d))I\} \geq 0$$

$$-\mathcal{S}(\rho) - (\log(1/d))\text{Tr}\{\rho\} \geq 0$$

Since $\text{Tr}\{\rho\} = 1$ and $\log(1/d) = -\log d$:

$$-\mathcal{S}(\rho) + \log d \geq 0 \implies \mathcal{S}(\rho) \leq \log d$$

The equality holds if and only if $\mathcal{S}(\rho||\sigma) = 0$, which implies $\rho = \sigma$. Therefore, the entropy is maximal and equal to $\log d$ if and only if the state is the completely mixed state, $\rho = \frac{1}{d}I$.

3. This property is a direct consequence of the Schmidt decomposition. Any pure state $|\Psi\rangle$ of a composite system $\mathcal{H}_A \otimes \mathcal{H}_B$ can be written as:

$$|\Psi\rangle = \sum_i \sqrt{\lambda_i} |u_i\rangle_A \otimes |v_i\rangle_B$$

where $\{|u_i\rangle_A\}$ and $\{|v_i\rangle_B\}$ are orthonormal sets in \mathcal{H}_A and \mathcal{H}_B respectively, and $\lambda_i > 0$ with $\sum_i \lambda_i = 1$.

The reduced density matrix for subsystem A is $\rho_A = \text{Tr}_B(|\Psi\rangle\langle\Psi|)$.

$$\rho_A = \text{Tr}_B \left(\sum_{i,j} \sqrt{\lambda_i \lambda_j} |u_i\rangle_A \langle u_j|_A \otimes |v_i\rangle_B \langle v_j|_B \right)$$

$$\rho_A = \sum_{i,j} \sqrt{\lambda_i \lambda_j} |u_i\rangle_A \langle u_j|_A \text{Tr}(|v_i\rangle_B \langle v_j|_B)$$

Since $\text{Tr}(|v_i\rangle\langle v_j|) = \langle v_j|v_i\rangle = \delta_{ij}$, we get:

$$\rho_A = \sum_i \lambda_i |u_i\rangle_A \langle u_i|_A$$

The non-zero eigenvalues of ρ_A are precisely the coefficients $\{\lambda_i\}$. The entropy is $\mathcal{S}(A) = -\sum_i \lambda_i \log \lambda_i$.

Similarly, the reduced density matrix for subsystem B is $\rho_B = \text{Tr}_A(|\Psi\rangle\langle\Psi|)$.

$$\rho_B = \sum_i \lambda_i |v_i\rangle_B \langle v_i|_B$$

The non-zero eigenvalues of ρ_B are also $\{\lambda_i\}$. The entropy is $\mathcal{S}(B) = -\sum_i \lambda_i \log \lambda_i$. Thus, $\mathcal{S}(A) = \mathcal{S}(B)$.

4. The equality $\mathcal{S}(\sum_i p_i \rho_i) = H(p_i) + \sum_i p_i \mathcal{S}(\rho_i)$ holds under the specific condition that the density matrices ρ_i have orthogonal support. This means that the vector spaces on which each ρ_i acts non-trivially are mutually orthogonal.

Let this condition hold. We can choose a basis for the total Hilbert space that respects this block structure. In this basis, the total density matrix $\rho = \sum_i p_i \rho_i$ is block-diagonal:

$$\rho = \begin{pmatrix} p_1 \rho_1 & 0 & \cdots \\ 0 & p_2 \rho_2 & \cdots \\ \vdots & \vdots & \ddots \end{pmatrix}$$

The set of eigenvalues of ρ is the union of the sets of eigenvalues of each block $p_i \rho_i$. If $\{\lambda_{ij}\}_j$ are the eigenvalues of ρ_i , then $\{p_i \lambda_{ij}\}_j$ are the eigenvalues of the i -th block.

The von Neumann entropy of ρ is the sum over all its eigenvalues:

$$\mathcal{S}(\rho) = -\sum_{i,j} (p_i \lambda_{ij}) \log(p_i \lambda_{ij})$$

Using the property of logarithms, $\log(ab) = \log a + \log b$:

$$\mathcal{S}(\rho) = -\sum_{i,j} p_i \lambda_{ij} (\log p_i + \log \lambda_{ij})$$

$$\mathcal{S}(\rho) = -\sum_{i,j} p_i \lambda_{ij} \log p_i - \sum_{i,j} p_i \lambda_{ij} \log \lambda_{ij}$$

We can split this into two parts. For the first part:

$$-\sum_{i,j} p_i \lambda_{ij} \log p_i = -\sum_i (p_i \log p_i) \left(\sum_j \lambda_{ij} \right)$$

Since $\sum_j \lambda_{ij} = \text{Tr}(\rho_i) = 1$, this simplifies to:

$$-\sum_i p_i \log p_i = H(p_i)$$

For the second part:

$$-\sum_{i,j} p_i \lambda_{ij} \log \lambda_{ij} = \sum_i p_i \left(-\sum_j \lambda_{ij} \log \lambda_{ij} \right)$$

The term in the parenthesis is the entropy of ρ_i , $\mathcal{S}(\rho_i)$. So this part becomes:

$$\sum_i p_i \mathcal{S}(\rho_i)$$

Combining the two parts gives the desired result:

$$\mathcal{S} \left(\sum_i p_i \rho_i \right) = H(p_i) + \sum_i p_i \mathcal{S}(\rho_i)$$

■

10.4 Quantum Data Compression

As a natural extension to Shannon's noiseless coding theorem, we have the quantum analogue presented below. For a message transmission of n letters, each letter being chosen at random from the alphabet \mathcal{A} , which here is an ensemble of pure states, defined by

$$\mathcal{A} = \{|\psi_1\rangle, |\psi_2\rangle, \dots, |\psi_k\rangle\}$$

The state $|\psi_i\rangle$ is extracted *a priori* with probability p_i , such that $\sum_{i=1}^k p_i = 1$. Thereby, for each letter in the message, we have the density matrix

$$\rho = \sum_{i=1}^k p_i |\psi_i\rangle \langle \psi_i|$$

thereby, for the entire message, we have the tensor product,

$$\rho_{(n)} = \rho^{\otimes n}$$

Here we have assumed that all the letters in the message are statistically independent and described by the same density matrix ρ .

Schumacher's theorem, similar to Shannon's coding theorem, entails us the machinery to encode the message, in the sense that we can compress the data, with the optimal compression rate directed by the von Neumann entropy.

Theorem 10.4.1. (*Schumacher's quantum noiseless coding theorem*) Suppose we have a message whose letters are drawn independently from the ensemble $\mathcal{A} = \{|\psi_1\rangle, \dots, |\psi_k\rangle\}$ with prior probabilities $\{p_1, \dots, p_k\}$, there exists an optimal and reliable code compressing the message to $\mathcal{S}(\rho)$ qubits per letter where $\rho = \sum_{i=1}^k p_i |\psi_i\rangle \langle \psi_i|$, asymptotically in the length of the message.

Proof. The proof of this theorem closely resembles the techniques used in the proof of Shannon's noiseless coding theorem described earlier¹. We illustrate the idea here, by spectrally decomposing the density operator ρ as

$$\rho = \sum_{i=1}^k \lambda_i |a_i\rangle\langle a_i|,$$

Further, we have the von Neuman entropy relating the classical optimal compression rate, as

$$H(\lambda_1, \dots, \lambda_k) = -\sum_i \lambda_i \log \lambda_i = -\text{Tr} \rho \log \rho = \mathcal{S}(\rho)$$

The ensemble $\tilde{\mathcal{A}}$ defined from the spectral decomposition states $\{|a_1\rangle, \dots, |a_k\rangle\}$ constitutes an alphabet of orthogonal pure quantum states.

We rework the definition of a ϵ -typical sequence, for a state $|x_1\rangle \otimes \dots \otimes |x_n\rangle$, with $|x_i\rangle \in \tilde{\mathcal{A}}$ is ϵ -typical, when

$$\left| -\frac{1}{n} \log[\lambda(x_1) \dots \lambda(x_n)] - \mathcal{S}(\rho) \right| < \epsilon$$

where $\lambda(x_i) = \lambda_j$ if $|x_i\rangle$ is in the letter $|a_j\rangle$. We define the ϵ -typical subspace as the subspace spanned by the ϵ -typical states.

As before, the dimension of the subspace can be shown to be of the order of $2^{n\mathcal{S}(\rho)}$. For any projector Π_{typical} on this typical subspace, we have

$$\text{Tr}\{\Pi_{\text{typical}}\rho^n\} > 1 - \delta$$

provided asymptotically large n , as we proved for the compression scheme in the classical case. Therefore, as $n \rightarrow \infty$, the density matrix $\rho_{(n)}$ has its support on a typical subspace of dimension $2^{n\mathcal{S}(\rho)}$. A typical n state message can then be encoded using $n\mathcal{S}(\rho)$ qubits, thus constraining the optimal rate by the von Neumann entropy. ■

Compression of an n qubit message

Consider the binary alphabet $\mathcal{A} = \{|\psi_0\rangle, |\psi_1\rangle\}$, where $|\psi_0\rangle = \cos \theta |0\rangle + \sin \theta |1\rangle$, and $|\psi_1\rangle = \sin \theta |0\rangle + \cos \theta |1\rangle$, which are not necessarily orthogonal.

Say we want to transfer the n qubit message,

$$|\Psi_K\rangle = |\psi_{k_1}\rangle \otimes |\psi_{k_2}\rangle \otimes \dots \otimes |\psi_{k_n}\rangle$$

where $K = \{k_1, \dots, k_n\}$ singles out the message, which each k_i being either 0 or 1.

The states $|\psi_0\rangle$ and $|\psi_1\rangle$ are drawn from the alphabet \mathcal{A} with probabilities p and $1-p$ respectively. Any n letter message $|\Psi_K\rangle$ is in the combined Hilbert space $\mathcal{H}^{\otimes n}$, for the Hilbert space \mathcal{H} of a single qubit.

¹Interested readers can refer to [Schumacher \[1995\]](#).

We decompose the message into the typical subspace through a projector as used in the proof, such that we can express

$$|\Psi_K\rangle = \alpha_K |\tau_K\rangle + \beta_K |\tau_K^\perp\rangle$$

where we say $|\tau_K\rangle$ to belong to the typical subspace $\mathcal{H}_{\text{typical}}$, and $|\tau_K^\perp\rangle$ belongs to the orthogonal complement space.

For a measurement to determine whether $|\Psi_K\rangle$ belongs to the typical subspace, such that the message is encoded, we realise that we need only $n\mathcal{S}(\rho)$ wubits for encoding, since the typical subspace has dimension $\approx 2^{n\mathcal{S}(\rho)}$. If instead $|\Psi_K\rangle$ belongs to the atypical subspace (given by the orthogonal complement), we substitute it with some reference state $|\mathcal{R}\rangle$ residing in the typical subspace.

On decoding the $n\mathcal{S}(\rho)$ qubits, we have the effective density matrix, given by

$$\tilde{\rho}_K = |\alpha_K|^2 |\tau_K\rangle\langle\tau_K| + |\beta_K|^2 |\mathcal{R}\rangle\langle\mathcal{R}|$$

As evidently seen, there is some notion of loss of information through the reference state. We can compute the effective reliability of compression through a physical quantity, termed the *fidelity* \mathcal{F} , given by

$$\mathcal{F} = \langle\Psi_K|\tilde{\rho}_K|\Psi_K\rangle$$

where we can clearly see that, if optimal compression, $\tilde{\rho} = |\Psi_K\rangle\langle\Psi_K|$, and we have $\mathcal{F} = 1$. If we have orthogonal initial and final states, then the fidelity vanishes, $\mathcal{F} = 0$.

We obtain the average fidelity $\bar{\mathcal{F}}$ by weighting over the probability of occurrence of the possible messages, such that, we have

$$\begin{aligned}\bar{\mathcal{F}} &= \sum_K p_K \langle\Psi_K|\tilde{\rho}_K|\Psi_K\rangle \\ &= \sum_K p_K \langle\Psi_K| (|\alpha_K|^2 |\tau_K\rangle\langle\tau_K| + |\beta_K|^2 |\mathcal{R}\rangle\langle\mathcal{R}|) |\Psi_K\rangle \\ &= \sum_K p_K |\alpha_K|^4 + \sum_K |\beta_K|^2 (|\langle\Psi_K|\mathcal{R}\rangle|^2)\end{aligned}$$

Thereby, we have average fidelity tending close to 1 as $n \rightarrow \infty$, such that messages overlap with the typical subspace. Hence, we can code only the typical subspace and still achieve good fidelity.

Further Reading & References

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Chapter 11

Exploiting Quantum Entanglement

“Feeling insignificant because the universe is large has exactly the same logic as feeling inadequate for not being a cow. Or a herd of cows. The universe is not there to overwhelm us; it is our home, and our resource. The bigger the better.”

– David Deutsch, *The Beginning of Infinity*

11.1 Introduction

The superposition principle illustrates the existence of entangled states in two or more quantum systems. These entangled states are characterised by cross-correlations between the systems, which any classical theory cannot satisfactorily explain. Such phenomena have played a central role in the development of quantum theory, beginning with the famous paradox posed by Einstein, Podolsky, and Rosen (EPR) and followed by the fascinating work of John Stewart Bell. This paradox exemplifies the seemingly absurd implications of entanglement when applied to the macroscopic world. The EPR dilemma challenges classical reasoning by presenting a conflict between the reality of physical properties and the locality implied by the finite speed of light. This challenge, along with subsequent developments, has refined our understanding of entanglement. In the field of quantum information, entanglement is considered a valuable resource to be utilised.

11.2 Local Operations Classical Communication

Say we play a game of quantum state exchange, starting with an entangled pure state $|\psi\rangle$ between us. Suppose we perform arbitrary operations on our local systems and can only communicate using classical communication channels. This exploration closely links with ideas of entanglement and a measure to quantify it through the different possible entanglement states $|\varphi\rangle$ it can transform into. These types of operations with intrinsic richness in

the class of transformations correspond to the class of *local operations and classical communication* (LOCC), which help us disentangle the ideas of bipartite quantum entanglement.

Quantum Teleportation

Quantum teleportation is an important task that can be completed by LOCC. Following the convention, this process requires 2 communication nodes or parties, namely A (Alice) and B (Bob). For simplicity, we only consider transferring a single-qubit quantum state $|\psi\rangle_C$ and this requires 3 qubits in total including the pre-shared maximally entangled state $|\Phi^+\rangle_{AB} = \frac{1}{\sqrt{2}}(|00\rangle + |11\rangle)$. Alice holds systems A and C , Bob holds system B . Note that only quantum information is transferred, not the physical qubits. The workflow proceeds in the following steps:

1. At the very beginning, the system state can be described as $|\varphi_0\rangle = |\psi\rangle_C \otimes |\Phi^+\rangle_{AB} = \frac{1}{\sqrt{2}}[\alpha|0\rangle(|00\rangle + |11\rangle) + \beta|1\rangle(|00\rangle + |11\rangle)]$ where the quantum state Alice want to transmit is $|\psi\rangle_C = \alpha|0\rangle_C + \beta|1\rangle_C$ and the coefficients $\alpha, \beta \in \mathbb{C}$.
2. Alice applies a CNOT gate, and the resulting state $|\varphi_1\rangle = \frac{1}{\sqrt{2}}[\alpha|0\rangle(|00\rangle + |11\rangle) + \beta|1\rangle(|10\rangle + |01\rangle)]$
3. Alice applies a Hadamard gate, and the system state becomes $|\varphi_2\rangle = \frac{1}{2}[\alpha(|0\rangle + |1\rangle)(|00\rangle + |11\rangle) + \beta(|0\rangle - |1\rangle)(|10\rangle + |01\rangle)]$. The above state can be rearranged to $|\varphi_2\rangle = \frac{1}{2}[|00\rangle(\alpha|0\rangle + \beta|1\rangle) + |01\rangle(\alpha|1\rangle + \beta|0\rangle) + |10\rangle(\alpha|0\rangle - \beta|1\rangle) + |11\rangle(\alpha|1\rangle - \beta|0\rangle)]$.
4. Alice measures both of her qubits in the computational basis $\{|00\rangle, |01\rangle, |10\rangle, |11\rangle\}$ and send the results m_1m_2 to Bob with a classical channel. There are 4 distinct possibilities: $m_1m_2 \in \{00, 01, 10, 11\}$. Then, Bob implements certain operations correspondingly on his qubit based on the received messages.
 - If the measurement result is $m_1m_2 = 00$, Bob's state will be $\alpha|0\rangle + \beta|1\rangle$, which is the state Alice want to transmit $|\psi\rangle_C$. No operations are needed and the teleportation is finished.
 - If the measurement result is $m_1m_2 = 01$, Bob's state will be $\alpha|1\rangle + \beta|0\rangle$. Bob needs to act the X gate on his qubit.
 - If the measurement result is $m_1m_2 = 10$, Bob's state will be $\alpha|0\rangle - \beta|1\rangle$. Bob needs to act the Z gate on his qubit.
 - If the measurement result is $m_1m_2 = 11$, Bob's state will be $\alpha|1\rangle - \beta|0\rangle$. Bob needs to act the X gate followed by the Z gate on his qubit.

In short, LOCC transfers quantum information between two spatially separated communication nodes (only a classical communication channel is allowed) with the help of entanglement. At the heart of entanglement theory is the notion of LOCC, since global quantum operations are unfeasible in regions separated physically.

We formalise the notion of LOCCs through the following theorem.

Theorem 11.2.1. *Let the state $|\varphi\rangle$ be transformed to $|\psi\rangle$ through the virtue of local operations and classical communication. This transformation expects a series of generalised measurement operators $\{M_i^A\}$ in virtue of A , transferring the measurement to B , who can transform the state by a pre-assigned unitary U_i to respect the change.*

Proof. Say that B performs a measurement with generalised measurement operators M_j^B on a pure state $|\varphi\rangle$. Let this state be Schmidt-decomposed as

$$|\varphi\rangle = \sum_{l=1}^{\mathcal{R}} \sqrt{\lambda_l} |l_A\rangle |l_B\rangle$$

with the Schmidt decomposition with $\text{rank}(\rho) = \mathcal{R}$. In this basis, we can define

$$M_j^B = \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} M_{j,kl} |k_B\rangle \langle l_B|$$

and we denote the specialized operator for A which is the same as the matrix representation with respect to A 's Schmidt basis as

$$M_j^A \equiv \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} M_{j,kl} |k_A\rangle \langle l_A|.$$

Now let B perform the measurement defined by these operators M_j^B , with post measurement state defined by

$$\begin{aligned} |\psi_j^B\rangle \propto M_j^B |\varphi\rangle &= \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} M_{j,kl} |k_B\rangle \langle l_B| |\varphi\rangle \\ &= \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} M_{j,kl} |k_B\rangle \langle l_B| \left(\sum_{l'=1}^{\mathcal{R}} \sqrt{\lambda_{l'}} |l'\rangle |l'_B\rangle \right) \\ &= \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} M_{j,kl} \sqrt{\lambda_l} |l_A\rangle |k_B\rangle. \end{aligned}$$

The probability of measurement is given by the norm as

$$\begin{aligned} p^B(j) = \|M_j^B |\varphi\rangle\|^2 &= \left(\sum_{l'}^{\mathcal{R}} \sum_{k'}^{\mathcal{R}} M_{j,k'l'}^* \sqrt{\lambda_{l'}} \langle l'_A| \langle k'_B| \right) \left(\sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} M_{j,kl} \sqrt{\lambda_l} |l_A\rangle |k_B\rangle \right) \\ &= \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} |M_{j,kl}|^2 \lambda_l, \end{aligned}$$

since we have the states $|l_A\rangle |k_B\rangle$ orthonormal in the Schmidt basis. Similarly, for A , we have the post-measurement state given by

$$|\psi_j^A\rangle \propto M_j^A |\varphi\rangle = \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} M_{j,kl} |k_A\rangle \langle l_A| |\varphi\rangle$$

$$\begin{aligned}
&= \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} M_{j,kl} |k_A\rangle \langle l_A| \left(\sum_{l'=1}^{\mathcal{R}} \sqrt{\lambda_{l'_A}} |l'\rangle \langle l'_B| \right) \\
&= \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} M_{j,kl} \sqrt{\lambda_l} |k_A\rangle \langle l_B|.
\end{aligned}$$

The probability of measurement is given by the norm as

$$\begin{aligned}
p^A(j) &= ||M_j^A|\varphi\rangle||^2 = \left(\sum_{l'}^{\mathcal{R}} \sum_{k'}^{\mathcal{R}} M_{j,k'l'}^* \sqrt{\lambda_{l'}} \langle k'_A | \langle l'_B| \right) \left(\sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} M_{j,kl} \sqrt{\lambda_l} |k_A\rangle \langle l_B| \right) \\
&= \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} |M_{j,kl}|^2 \lambda_l,
\end{aligned}$$

since we have the states $|k_A\rangle \langle l_B|$ orthonormal in the Schmidt basis.

Thereby, the probabilities are inherently equivalent $p^A(j) = p^B(j)$ and we have the states $|\psi_j^A\rangle$ and $|\psi_j^B\rangle$ are related by an unitary transformation admitting the change of basis from $|l_A\rangle \langle k_B|$ to $|k_A\rangle \langle l_B|$ as

$$\begin{aligned}
\psi_j^B &= (U_j^A \otimes V_j^B) \psi_j^A \\
&= (U_j^A \otimes V_j^B) \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} M_{j,kl} \sqrt{\lambda_l} |k_A\rangle \langle l_B| \\
&= \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} M_{j,kl} \sqrt{\lambda_l} (U_j^A |k_A\rangle) (V_j^B \langle l_B|),
\end{aligned}$$

such that $U_j^A |k_A\rangle = |l_A\rangle$ and $V_j^B \langle k_B| = \langle l_B|$.

Therefore, B performing a measurement described by measurement operators M_j is equivalent to A performing the measurement described by measurement operators $U_j^A M_j^A$ followed by B performing the unitary transformation V_j^B . In summarising, a measurement by B on a known pure state can be simulated by a measurement by A , up to a unitary transformation by B . ■

Further, we note the resulting post-measurement density matrix due to B 's measurement given by

$$\begin{aligned}
\rho' &= \frac{M_j^B \rho M_j^{B\dagger}}{\text{Tr}(\rho M_j^{B\dagger} M_j^B)} \propto M_j^B |\varphi\rangle \langle \varphi| M_j^{B\dagger} \\
&= \left(\sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} M_{j,kl} \sqrt{\lambda_l} |l_A\rangle \langle k_B| \right) \left(\sum_{l'}^{\mathcal{R}} \sum_{k'}^{\mathcal{R}} M_{j,k'l'}^* \sqrt{\lambda_{l'}} \langle l'_A| \langle k'_B| \right) \\
&= \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} \sum_{l'}^{\mathcal{R}} \sum_{k'}^{\mathcal{R}} M_{j,kl} M_{j,k'l'}^* \sqrt{\lambda_l \lambda_{l'}} |l_A\rangle \langle l'_A| |k_B\rangle \langle k'_B|,
\end{aligned}$$

which implies, the reduced density matrices are

$$\begin{aligned}\rho_j'^A &= \text{Tr}_B(\rho_j') = \sum_l^{\mathcal{R}} \sum_{l'}^{\mathcal{R}} \sum_k^{\mathcal{R}} M_{j,kl} M_{j,kl'}^* \sqrt{\lambda_l \lambda_{l'}} |l_A\rangle \langle l'_A|, \\ \rho_j'^B &= \text{Tr}_A(\rho_j') = \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} \sum_{k'}^{\mathcal{R}} M_{j,kl} M_{j,k'l}^* \lambda_l |k_B\rangle \langle k'_B|.\end{aligned}$$

Further, due to A 's measurements, we have

$$\begin{aligned}\rho_j'' &= \frac{M_j^A \rho_j^{A\dagger}}{\text{Tr}(\rho_j^{A\dagger} M_j^A)} \propto M_j^A |\varphi\rangle \langle \varphi| M_j^{A\dagger} \\ &= \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} \sum_{l'}^{\mathcal{R}} \sum_{k'}^{\mathcal{R}} M_{j,kl} M_{j,k'l'}^* \sqrt{\lambda_l \lambda_{l'}} |k_A\rangle \langle k'_A| |l_B\rangle \langle l'_B|,\end{aligned}$$

which implies, the reduced density matrices are

$$\begin{aligned}\rho_j''^A &= \text{Tr}_B(\rho_j'') = \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} \sum_{k'}^{\mathcal{R}} M_{j,kl} M_{j,k'l}^* \lambda_l |k_A\rangle \langle k'_A|, \\ \rho_j''^B &= \text{Tr}_A(\rho_j'') = \sum_l^{\mathcal{R}} \sum_{l'}^{\mathcal{R}} \sum_k^{\mathcal{R}} M_{j,kl} M_{j,kl'}^* \sqrt{\lambda_l \lambda_{l'}} |l_B\rangle \langle l'_B|.\end{aligned}$$

We note that ρ_j'' and ρ_j'' can be related by the change of basis matrices as before by the transformation

$$\begin{aligned}\rho_j' &= (U_j^A \otimes V_j^B) \rho_j'' (U_j^{A\dagger} \otimes V_j^{B\dagger}) \\ &= (U_j^A \otimes V_j^B) \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} \sum_{l'}^{\mathcal{R}} \sum_{k'}^{\mathcal{R}} M_{j,kl} M_{j,k'l'}^* \sqrt{\lambda_l \lambda_{l'}} |k_A\rangle \langle k'_A| |l_B\rangle \langle l'_B| (U_j^{A\dagger} \otimes V_j^{B\dagger}) \\ &= \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} \sum_{l'}^{\mathcal{R}} \sum_{k'}^{\mathcal{R}} M_{j,kl} M_{j,k'l'}^* \sqrt{\lambda_l \lambda_{l'}} U_j^A |k_A\rangle \langle k'_A| U_j^{A\dagger} V_j^B |l_B\rangle \langle l'_B| V_j^{B\dagger}.\end{aligned}$$

such that $U_j^A |k_A\rangle = |l_A\rangle$ and $V_j^B |k_B\rangle = |l_B\rangle$. Note that the same does not hold for $\rho_j'^A$ and $\rho_j'^B$ defined by the partial trace. In fact, we have $\rho_j'^A = \rho_j''^B$ and $\rho_j'^B = \rho_j''^A$. Further, we note

$$\begin{aligned}U_j^A \rho_j''^A U_j^{A\dagger} &= \sum_l^{\mathcal{R}} \sum_k^{\mathcal{R}} \sum_{k'}^{\mathcal{R}} M_{j,kl} M_{j,k'l}^* \lambda_l U_j^A |k_A\rangle \langle k'_A| U_j^{A\dagger} \neq \rho_j'^A \\ V_j^B \rho_j''^B V_j^{B\dagger} &= \sum_l^{\mathcal{R}} \sum_{l'}^{\mathcal{R}} \sum_k^{\mathcal{R}} M_{j,kl} M_{j,kl'}^* \sqrt{\lambda_l \lambda_{l'}} V_j^B |l_B\rangle \langle l'_B| V_j^{B\dagger} \neq \rho_j'^B.\end{aligned}$$

11.3 Majorization

Majorization is a purely mathematical concept with surprisingly far-reaching applications. Consider two vectors $\mathbf{x}, \mathbf{y} \in \mathbb{R}^n$, where we define a sorted (in non-ascending manner) version of a vector \mathbf{a} as \mathbf{a}^\downarrow , such that

$$a_1^\downarrow \geq a_2^\downarrow \geq \cdots \geq a_n^\downarrow$$

Note that, the sorted vector is a permutation of the elements of \mathbf{v} , hence we can relate the entries of the descending vector through a permutation matrix $\mathcal{P} \in S_n$, such that

$$\mathbf{v}^\downarrow = \mathcal{P}\mathbf{v}, \quad \mathcal{P} \in S_n$$

We define \mathbf{x} *majorizes* \mathbf{y} , written as $\mathbf{x} \succ \mathbf{y}$, if

$$\mathbf{x} \succ \mathbf{y} \implies \sum_{j=1}^k x_j^\downarrow = \sum_{j=1}^k y_j^\downarrow \quad \forall 1 \leq k \leq n$$

The central insight into majorization theory relies on the idea that

$$\mathbf{x} \succ \mathbf{y} \iff \mathbf{y} = \sum_j p_j \mathcal{P}_j \mathbf{x}$$

for a probability distribution p_j over the permutation matrices \mathcal{P}_j . This can be understood through the inductive reasoning that, for $\mathbf{x} \succ \mathbf{y}$, the biggest element of \mathbf{x}^\downarrow must exceed the last element of \mathbf{y}^\downarrow and the difference of their sums, such that a convex combination of \mathbf{x} 's are obtained for \mathbf{y} . Thus, $\mathbf{x} \succ \mathbf{y}$ if and only if \mathbf{y} can be written as a convex combination of permutations of \mathbf{x} , resulting in a more disordered sense and intermixing the elements of the vectors.

These matrices, written as a convex combination of permutation matrices, give rise to rich physical insight. The entries of these matrices are non-negative, and the sums of columns and rows are identity. Through the implications of Birkhoff's phenomenal theorem¹, we can rewrite

$$\mathbf{y} = \mathcal{D}\mathbf{x}$$

where \mathcal{D} is doubly-stochastic, which has all columns and rows as simultaneously probability distributions, that is

$$\mathcal{D}_{ij} \geq 0, \quad \sum_{i=1}^n \mathcal{D}_{ij} = \sum_{j=1}^n \mathcal{D}_{ij} = 1$$

11.4 Entanglement Transformations

Through the ideas of majorization, we can uncover the intricate aspects of quantum entanglement by understanding when we can transform a given copy of a pure bipartite quantum

¹The proof of this theorem involves beautiful implications of graph theory and mapping doubly stochastic matrices to an associated graph, further using a very beautiful concept involving Hall's marriage theorem. Refer [Hetyei \[2016\]](#).

state $|\psi\rangle$ to another quantum state $|\varphi\rangle$ using LOCC². Symbolically, we shall investigate

$$|\psi\rangle \xrightarrow{\text{LOCC}} |\varphi\rangle$$

As a first ingredient, we shall extend the definition of majorization to general density matrices that are Hermitian. We define majorization of $\rho_\psi = \text{Tr}_B\{|\psi\rangle\langle\psi|\}$ and $\rho_\varphi = \text{Tr}_B\{|\varphi\rangle\langle\varphi|\}$ such that

$$\rho_\psi \succ \rho_\varphi \iff \boldsymbol{\lambda}_\psi \succ \boldsymbol{\lambda}_\varphi$$

where $\boldsymbol{\lambda}$ is a vector containing the eigenvalues.

From the above analogy of doubly stochastic matrices, we proceed to prove that $\rho_\psi \succ \rho_\varphi$ if and only if we have a stochastic unitary transformation of ρ_ψ to ρ_φ .

Theorem 11.4.1. *For Hermitian operators ρ_ψ, ρ_φ , we have $\rho_\psi \succ \rho_\varphi$ if and only if there exists a probability distribution p_j and unitary matrices U_j such that*

$$\rho_\varphi = \sum_j p_j U_j \rho_\psi U_j^\dagger$$

Proof. (\implies): Let ρ_ψ, ρ_φ . By definition, $\rho_\psi \succ \rho_\varphi$ implies $\boldsymbol{\lambda}_\psi \succ \boldsymbol{\lambda}_\varphi$, hence there exists a convex combination transformation through permutation matrix $\mathcal{P}_j \in S_n$ from the above proposition such that

$$\boldsymbol{\lambda}_\varphi = \sum_j p_j \mathcal{P}_j \boldsymbol{\lambda}_\psi$$

To transform from the eigenvalues to the density matrix, consider the diagonalisations through unitary transformations

$$\rho_\psi = \mathcal{S}_\psi^\dagger \Lambda_\psi \mathcal{S}_\psi, \quad \rho_\varphi = \mathcal{S}_\varphi^\dagger \Lambda_\varphi \mathcal{S}_\varphi$$

Now, note that the vectorial equation $\boldsymbol{\lambda}_\varphi = \sum_j p_j \mathcal{P}_j \boldsymbol{\lambda}_\psi$ can be expressed as

$$\Lambda_\varphi = \sum_j p_j \mathcal{P}_j \Lambda_\psi \mathcal{P}_j^\dagger$$

Through the inverse transformation, we recover the density matrices

$$\begin{aligned} \rho_\varphi &= \mathcal{S}_\varphi \Lambda_\varphi \mathcal{S}_\varphi^\dagger = \mathcal{S}_\varphi \left(\sum_j p_j \mathcal{P}_j \Lambda_\psi \mathcal{P}_j^\dagger \right) \mathcal{S}_\varphi^\dagger \\ &= \sum_j p_j \mathcal{S}_\varphi \mathcal{P}_j \Lambda_\psi \mathcal{P}_j^\dagger \mathcal{S}_\varphi^\dagger = \sum_j p_j \mathcal{S}_\varphi \mathcal{P}_j (\mathcal{S}_\psi^\dagger \rho_\psi \mathcal{S}_\psi) \mathcal{P}_j^\dagger \mathcal{S}_\varphi^\dagger \\ &= \sum_j p_j (\mathcal{S}_\varphi \mathcal{P}_j \mathcal{S}_\psi^\dagger) \rho_\psi (\mathcal{S}_\psi \mathcal{P}_j^\dagger \mathcal{S}_\varphi^\dagger) \equiv \sum_j p_j \tilde{\mathcal{P}}_j \rho_\psi \tilde{\mathcal{P}}_j^\dagger \end{aligned}$$

²The original idea explaining what tasks may be accomplished using a given physical resource and the ideas for entanglement transformations was first presented by Michael Nielsen in [Nielsen \[1999\]](#).

where we define $\tilde{\mathcal{P}}_j := \mathcal{S}_\varphi \mathcal{P}_j \mathcal{S}_\psi^\dagger$, such that $\tilde{\mathcal{P}}_j^\dagger = \mathcal{S}_\psi \mathcal{P}_j^\dagger \mathcal{S}_\varphi^\dagger$. Note that the composition of unitary matrices with a permutation matrix, still results in another permutation matrix. We have completed the proof in the forward direction.

(\Leftarrow): Let $\rho_\varphi = \sum_j p_j U_j \rho_\psi U_j^\dagger$. On diagonalising, we have

$$\begin{aligned} \Lambda_\varphi &= \mathcal{S}_\varphi^\dagger \rho_\varphi \mathcal{S}_\varphi = \mathcal{S}_\varphi^\dagger \left(\sum_j p_j U_j \rho_\psi U_j^\dagger \right) \mathcal{S}_\varphi \\ &= \sum_j p_j \mathcal{S}_\varphi^\dagger U_j (\mathcal{S}_\psi \Lambda_\psi \mathcal{S}_\psi^\dagger) U_j^\dagger \mathcal{S}_\varphi \\ &= \sum_j p_j (\mathcal{S}_\varphi^\dagger U_j \mathcal{S}_\psi) \Lambda_\psi (\mathcal{S}_\psi^\dagger U_j^\dagger \mathcal{S}_\varphi) \equiv \sum_j p_j V_j \Lambda_\psi V_j^\dagger \end{aligned}$$

where we define $V_j := \mathcal{S}_\varphi^\dagger U_j \mathcal{S}_\psi$ and subsequently, we have $V_j^\dagger = \mathcal{S}_\psi^\dagger U_j^\dagger \mathcal{S}_\varphi$, which are unitaries. Now, the matrix components can be identified for V_j as $V_{j,kl}$ such that we have

$$(\Lambda_\varphi)_k = \sum_{jl} p_j V_{j,kl} (\Lambda_\psi)_l V_{j,lk}^\dagger = \sum_{jl} p_j |V_{j,kl}|^2 (\Lambda_\psi)_l$$

We define a matrix \mathcal{D} with entries

$$\mathcal{D}_{kl} = \sum_j p_j |V_{j,kl}|^2$$

such that we have

$$\Lambda_\varphi = \mathcal{D} \Lambda_\psi$$

The entries of \mathcal{D} are non-negative by definition, and we have rows and columns summing up to unitary, thereby, the matrix \mathcal{D} is doubly stochastic and we have

$$\rho_\psi \succ \rho_\varphi$$

■

We can now proceed to characterising bipartite entanglement through the notion of majorization.

Theorem 11.4.2. *A bipartite pure state $|\varphi\rangle$ can be transformed to another pure state $|\psi\rangle$ by LOCC if and only if $\rho_\psi \succ \rho_\varphi$.*

Proof. (\Rightarrow): Suppose $|\varphi\rangle$ is transformed to state $|\psi\rangle$ by virtue of local operations and classical communication. By Theorem 11.2.1, we can assume that a bipartite system represented by A and B , with A performing a measurement with generalised measurement operators $\{M_i^A\}$, then sending the result to B , who performs an unitary transformation U_i . From the post-measurement theorem, we have A with density matrix ρ_φ transforming to state ρ_ψ , such that

$$\rho_\psi = \frac{M_j^A \rho_\varphi M_j^{A\dagger}}{\text{Tr}(\rho_\varphi M_j^{A\dagger} M_j^A)}$$

Further, to express ρ_φ as a convex combinations of elements of ρ_ψ , note that we could polar decompose the matrix $M_j^A \sqrt{\rho_\varphi}$, such that there exists a unitary V_j that

$$M_i^A \sqrt{\rho_\varphi} := \sqrt{M_i^A \rho_\varphi M_i^{A\dagger}} V_i = \sqrt{\text{Tr}(\rho_\varphi M_j^{A\dagger} M_j^A) \rho_\psi} V_i = \sqrt{p_i \rho_\psi} V_i$$

where p_i is the probability of outcome i . Premultiplying this equation by its adjoint, we thus realise,

$$\begin{aligned} (M_i^A \sqrt{\rho_\varphi})^\dagger M_i^A \sqrt{\rho_\varphi} &= (\sqrt{p_i \rho_\psi} V_i)^\dagger (\sqrt{p_i \rho_\psi} V_i) \\ \implies \sqrt{\rho_\varphi} M_i^{A\dagger} M_i^A \sqrt{\rho_\varphi} &= p_i V_i^\dagger \rho_\psi V_i \end{aligned}$$

Further, the completeness relation can be implemented for $\sum_i M_i^{A\dagger} M_i^A = \mathbb{I}$, such that

$$\begin{aligned} \sum_i p_i V_i^\dagger \rho_\psi V_i &= \sum_i \sqrt{\rho_\varphi} M_i^{A\dagger} M_i^A \sqrt{\rho_\varphi} \\ &= \sqrt{\rho_\varphi} \left(\sum_i M_i^{A\dagger} M_i^A \right) \sqrt{\rho_\varphi} = \rho_\varphi \end{aligned}$$

Hence, we have

$$\rho_\varphi = \sum_i p_i V_i^\dagger \rho_\psi V_i$$

and by Theorem 11.4.1, we can conclude $\rho_\psi \succ \rho_\varphi$

(\Leftarrow): Let us assume $\rho_\psi \succ \rho_\varphi$, then we can pose Theorem 11.4.1, for the existence of a probability distribution p_j and unitary matrices U_j such that

$$\rho_\varphi = \sum_i p_i U_i \rho_\psi U_i^\dagger$$

Motivated by the previous instance, we construct operators M_j^A through the action

$$M_i^A \sqrt{\rho_\varphi} := \sqrt{\text{Tr}(\rho_\varphi M_i^{A\dagger} M_i^A) \rho_\psi} U_i^\dagger = \sqrt{p_i \rho_\psi} U_i^\dagger$$

These define a set of measurement operators $\{M_i^A\}$ as seen from the completeness relation

$$\begin{aligned} \sum_i \sqrt{\rho_\varphi} M_i^{A\dagger} M_i^A \sqrt{\rho_\varphi} &= \sum_i (M_i^A \sqrt{\rho_\varphi})^\dagger (M_i^A \sqrt{\rho_\varphi}) \\ &= \sum_i (\sqrt{p_i \rho_\psi} U_i^\dagger)^\dagger (\sqrt{p_i \rho_\psi} U_i^\dagger) \\ &= \sum_i p_i U_i \rho_\psi U_i^\dagger = \rho_\varphi \end{aligned}$$

Hence, inverting the matrix $\sqrt{\rho_\varphi}$, we have

$$\sum_i M_i^{A\dagger} M_i^A = \rho_\varphi^{-\frac{1}{2}} \rho_\varphi \rho_\varphi^{-\frac{1}{2}} = \mathbb{I}$$

which proves the completeness relation. Thereby, A performs the measurement described by operators $\{M_i^A\}$, obtaining outcome i and corresponding state $|\psi_i^A\rangle \propto M_i^A|\varphi\rangle$. The reduced density matrix corresponding to the state $|\psi_i^A\rangle$ is $\rho_{\psi,i}^A = \text{Tr}_B\{|\psi_i^A\rangle\langle\psi_i^A|\}$, thereby

$$\begin{aligned}\rho_{\psi,i} &\propto \text{Tr}_B\{M_i^A|\varphi\rangle\langle\varphi|M_i^{A\dagger}\} \\ &= M_i^A\rho_\varphi M_i^{A\dagger} = (M_i^A\sqrt{\rho_\varphi})(M_i^A\sqrt{\rho_\varphi})^\dagger \\ &= (\sqrt{p_i\rho_\psi}U_i^\dagger)(\sqrt{p_i\rho_\psi}U_i^\dagger)^\dagger \\ &= p_i\sqrt{\rho_\psi}U_i^\dagger U_i\sqrt{\rho_\psi} = p_i\rho_\psi\end{aligned}$$

Hence, up to normalization $\rho_{\psi,i} \equiv \rho_\psi$. Now consider the state $|\psi_i^A\rangle$, which we can convert to $|\psi\rangle$ through the action of an unitary V_i such that the density matrices are equivalent. Thus, we can convert state $|\varphi\rangle$ to state $|\psi\rangle$ by virtue of LOCC. \blacksquare

Further Reading & References

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Chapter 12

Quantum Error Correction

“To err is human, to forgive divine.”

– Alexander Pope, *An Essay on Criticism*

12.1 Introduction

Any physical system is prone to errors. Say, suppose we isolate a quantum system from external decoherence, thus preventing errors from the environment. But, as quantum gates are unitary transformations chosen from a continuum of possible values, they cannot be implemented with perfect accuracy. Effects of tiny imperfections in the gate can accumulate and cause fatal errors. So we have a significant problem to tackle in quantum computing.

All the algorithms and the usefulness of quantum computers we saw in this text so far are not of practical utility if the errors cannot be corrected. So our next task is to identify and correct errors. Where do we start? Drawing inspiration from classical error correction, we can try to devise quantum error correction.

In the classical framework, we can try repetition to ensure redundancy. The most general classical single-bit error is the bit-flip $0 \leftrightarrow 1$. The simplest form of redundancy involves keeping multiple copies of each bit. For example, if we maintain two copies of each bit, a $\bar{0}$ is represented by the pair 00, while a $\bar{1}$ is represented by the pair 11. If an error occurs in one of the bits, we might end up with the pairs 01 or 10. Since these pairs should never occur, encountering them serves as an indication that an error has taken place.

Slightly more technically, the strings 00 and 11 have *even parity* and if we detect a string with odd parity, we know that an error has occurred. This is an *error-detecting code*. But, not only do we want to detect, but we want to correct errors too. We can do that by increasing redundancy and keeping 3 copies of each bit: $\bar{0} \rightarrow 000$, and $\bar{1} \rightarrow 111$. If an error occurs, we get one of the strings 001, 010, 100, 110, 101, 011. In this case, we correct the bit by using the *majority* value, with the rule that the transformation selects the maximum

occurring value, either 0 or 1. Thereby, we have to check for the value that appears maximally in the 3 bit string.

Now, a similar act of error correction on a quantum computer seems to be extremely non-trivial, due to

1. *Lack of Repetition:* To note if there are errors, we need to inherently measure the state, which would imply the collapse of the state, and we are restricted by the no-cloning nature of quantum mechanics from having copies to be redundant.
2. *Error Source:* Measuring a qubit to know the value can destroy its quantum correlations with other qubits with which it might be entangled. There could be other forms of errors too, including phase flip or other issues, which are purely quantum in nature.

Despite all these hurdles, there is an enormous conceptual progress and literature on quantum error correction, along with practical implementations, that are raising hopes for practically useful quantum computers in the near future. Our goal is to sketch Shor's original construction of a quantum error correcting code, and show how we avoid the conclusions that quantum error correction is impossible.

12.2 Essential features of Quantum Error Correction

We shall further build the idea of quantum error codes by studying the analogous classical error of flipping a qubit, followed by the intrinsic quantum channels. We aim to provide an independent description and also motivate the error channels under the Kraus operator and Stabilizer formalism that shall come further.

12.2.1 Bit Flip Code

Consider a case where the error channel flips $|0\rangle$ to $|1\rangle$ or the other way with probability p . Let us, for simplicity, consider a pure state, but the arguments are analogous for mixed states. Let $\rho = |\phi\rangle\langle\phi|$ be the initial density matrix of the system. Then, after passing through the error channel, we assume a noisy ensemble. It is helpful to view the bit-flip channel as an ensemble with probability $1 - p$ nothing happens, and with probability p an X flip occurs. It becomes

$$\rho_{\text{final}} = (1 - p) |\phi\rangle\langle\phi| + pX |\phi\rangle\langle\phi| X$$

It must be stressed that the above quantities can be thought of trajectory averaged quantum operators, of which, even if a specific realisation is picked, the formalism still holds strong. In any single experimental run, the system follows one definite trajectory; our formalism projects onto that trajectory, and the recovery is applied conditionally.

If we do not take any additional caution to correct the error, then the probability of loss of information is

$$P_{\text{error}} = 1 - \langle\phi| \rho_{\text{final}} |\phi\rangle.$$

If we consider the initial state to be a pure state $|\phi\rangle$, then,

$$P_{\text{error}} = 1 - (1 - p + p |\langle \phi | X | \phi \rangle|^2)$$

Thus, the probability of failure is of order p . Is there a way to reduce this error?

Having redundancy is a good old method to reduce errors, termed as *repetition code*. Even if an error occurs, it is less likely to affect all our redundant qubits. By majority rule, we can determine and correct the errors. We know that the no-cloning theorem says we cannot copy a quantum state, but instead we can copy the basis state. Although the no-cloning theorem forbids a single quantum operation that takes an *arbitrary* unknown state to two independent copies, it does *not* forbid copying the values of a known orthonormal basis. Concretely, a CNOT does copy computational-basis bits. Consider making three copies of the basis.

$$|\bar{0}\rangle \rightarrow |000\rangle \text{ and } |\bar{1}\rangle \rightarrow |111\rangle$$

These $|\bar{0}\rangle$ and $|\bar{1}\rangle$ are called *logical qubit states* and the ones without overline, $|0\rangle$ and $|1\rangle$ are *physical qubits*. Thus, we are *encoding* a single qubit state in a three-qubit Hilbert space.

In other words, our state $|\phi\rangle$ is encoded as,

$$|\phi\rangle = \alpha |0\rangle + \beta |1\rangle \longrightarrow |\bar{\psi}\rangle = \alpha |000\rangle + \beta |111\rangle$$

The circuit in the Fig. 12.1 does exactly this.

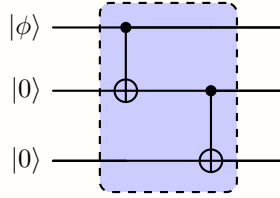


Figure 12.1: Copying the basis state

Algebraically, using projectors $P_0 = |0\rangle\langle 0|$ and $P_1 = |1\rangle\langle 1|$, the two CNOTs in the encoding circuit can be written as

$$\begin{aligned} \text{CNOT}_{12} &= (P_0^{(1)} \otimes \mathbb{I}^{(2)} + P_1^{(1)} \otimes X^{(2)}) \otimes \mathbb{I}^{(3)}, \\ \text{CNOT}_{23} &= \mathbb{I}^{(1)} \otimes (P_0^{(2)} \otimes \mathbb{I}^{(3)} + P_1^{(2)} \otimes X^{(3)}) \end{aligned}$$

where superscripts indicate on which qubit the operators act.

Start with the encoded input $|\phi\rangle = \alpha |0\rangle + \beta |1\rangle$ and two ancillas,

$$|\psi_{\text{in}}\rangle = (\alpha |0\rangle + \beta |1\rangle) \otimes |0\rangle \otimes |0\rangle = \alpha |000\rangle + \beta |100\rangle$$

Apply the first CNOT (control qubit 1, target qubit 2)

$$\text{CNOT}_{12} |\psi_{\text{in}}\rangle = \text{CNOT}_{12} (\alpha |000\rangle + \beta |100\rangle)$$

$$\begin{aligned}
&= \alpha(P_0^{(1)} \otimes \mathbb{I}^{(2)}) |000\rangle + \beta(P_1^{(1)} \otimes X^{(2)}) |100\rangle \\
&= \alpha |000\rangle + \beta |110\rangle.
\end{aligned}$$

Now apply the second CNOT (control qubit 2, target qubit 3)

$$\begin{aligned}
\text{CNOT}_{23}(\alpha |000\rangle + \beta |110\rangle) &= \alpha(P_0^{(2)} \otimes \mathbb{I}^{(3)}) |000\rangle + \beta(P_1^{(2)} \otimes X^{(3)}) |110\rangle \\
&= \alpha |000\rangle + \beta |111\rangle.
\end{aligned}$$

Thus, the complete encoding yields the expected logical state

$$|\bar{\psi}\rangle = \alpha |000\rangle + \beta |111\rangle.$$

Now, with three independently noisy physical qubits (each flipped with probability p), every subset $S \subset \{1, 2, 3\}$ of flipped qubits occurs as a trajectory with probability $p^{|S|}(1-p)^{3-|S|}$. If we write $X_S \equiv \prod_{i \in S} X_i$ (so $X_\emptyset = \mathbb{I}$, $X_{\{1,2\}} = X_1 X_2$, etc.), the output of the channel is the classical mixture over all flip patterns,

$$\sigma' = \sum_{S \subset \{1,2,3\}} p^{|S|}(1-p)^{3-|S|} X_S |\bar{\psi}\rangle \langle \bar{\psi}| X_S$$

Grouping terms by the number of flips gives the expanded form

$$\begin{aligned}
\sigma' &= (1-p)^3 |\bar{\psi}\rangle \langle \bar{\psi}| + p(1-p)^2 \sum_{i=1}^3 X_i |\bar{\psi}\rangle \langle \bar{\psi}| X_i \\
&\quad + p^2(1-p) \sum_{1 \leq i < j \leq 3} X_i X_j |\bar{\psi}\rangle \langle \bar{\psi}| X_i X_j + p^3 X_1 X_2 X_3 |\bar{\psi}\rangle \langle \bar{\psi}| X_1 X_2 X_3,
\end{aligned}$$

where the combinatorial factors are the binomial probabilities: choose which qubits flipped, each chosen flip contributes a factor p and each non-flip a factor $1-p$. This map can be shown to be trace-preserving.

Even though we encoded the logical state across three physical qubits, a direct projective measurement on any single physical qubit would reveal (and thus destroy) part of the logical superposition. The trick of error correction is to perform *collective* parity checks which commute with the logical operator and therefore reveal only which flip-pattern (if any) occurred, without collapsing the logical amplitudes in $|\bar{\psi}\rangle = \alpha |000\rangle + \beta |111\rangle$.

Consider the operators $Z_1 Z_2$ and $Z_2 Z_3$ and their action on the following states. The Table 12.1 shows some possible states the three qubits may be in after going through the error channel. Notice that both $Z_1 Z_2$ and $Z_2 Z_3$ do not change the state of any of these. In other words, the tabulated states are eigenstates of $Z_1 Z_2$ and $Z_2 Z_3$ whose eigenvalues are helping in locating the errors. Such measurements is called a *syndrome measurement*.

It is important that our measurement to diagnose the bit flip is a collective measurement on two qubits at once. We infer the value of $Z_1 Z_2$ and $Z_2 Z_3$ but get to learn nothing about the separate values of Z_1 , Z_2 or Z_3 , doing so would damage the encoded state.

State	Z_1Z_2	Z_2Z_3	Action to correct the error
$\alpha 000\rangle + \beta 111\rangle$	+1	+1	\mathbb{I}
$\alpha 100\rangle + \beta 011\rangle$	-1	+1	X_1
$\alpha 010\rangle + \beta 101\rangle$	-1	-1	X_2
$\alpha 001\rangle + \beta 110\rangle$	+1	-1	X_3
$\alpha 110\rangle + \beta 001\rangle$	+1	-1	X_3

Table 12.1: Action of Z_1Z_2 and Z_2Z_3 on various three-qubit states

We can perform the above pair of Z -measurements using the circuit 12.2 with the help of two additional *ancilla* qubits. Note that the two CNOTs outside the box is the one we saw earlier that copies the basis state.

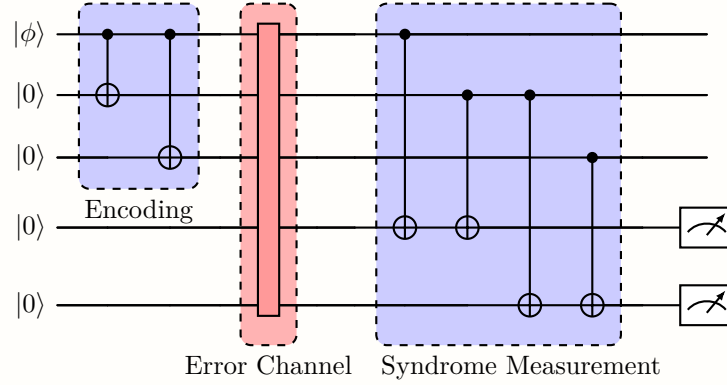


Figure 12.2: Syndrome measurements for bit flip error

After performing Z_1Z_2 and Z_2Z_3 we get 4 possible final states (syndromes) for all possible bit flip errors (i.e not only single bit flip).

Notice both Z_1Z_2 and Z_2Z_3 have +1 eigenvalue for the no-error state $\alpha|000\rangle + \beta|111\rangle$ and the three-flip error state $X_1X_2X_3|\bar{\psi}\rangle = \alpha|111\rangle + \beta|000\rangle$.

$$\begin{aligned}
\sigma_0 &= (1-p)^3|\bar{\psi}\rangle\langle\bar{\psi}| + p^3X_1X_2X_3|\bar{\psi}\rangle\langle\bar{\psi}|X_1X_2X_3 \\
\sigma_1 &= (1-p)^2pX_1|\bar{\psi}\rangle\langle\bar{\psi}|X_1 + (1-p)p^2X_2X_3|\bar{\psi}\rangle\langle\bar{\psi}|X_2X_3 \\
\sigma_2 &= (1-p)^2pX_2|\bar{\psi}\rangle\langle\bar{\psi}|X_2 + (1-p)p^2X_1X_3|\bar{\psi}\rangle\langle\bar{\psi}|X_1X_3 \\
\sigma_3 &= (1-p)^2pX_3|\bar{\psi}\rangle\langle\bar{\psi}|X_3 + (1-p)p^2X_1X_2|\bar{\psi}\rangle\langle\bar{\psi}|X_1X_2
\end{aligned}$$

Now depending on the Z_1Z_2, Z_2Z_3 values we operate with X_1, X_2 or X_3 . After which the state becomes

$$\begin{aligned}
\sigma'_0 &= (1-p)^3|\bar{\psi}\rangle\langle\bar{\psi}| + p^3X_1X_2X_3|\bar{\psi}\rangle\langle\bar{\psi}|X_1X_2X_3 \\
\sigma'_1 &= (1-p)^2p|\bar{\psi}\rangle\langle\bar{\psi}| + (1-p)p^2X_1X_2X_3|\bar{\psi}\rangle\langle\bar{\psi}|X_1X_2X_3
\end{aligned}$$

$$\begin{aligned}\sigma'_2 &= (1-p)^2 p |\bar{\psi}\rangle\langle\bar{\psi}| + (1-p)p^2 X_1 X_2 X_3 |\bar{\psi}\rangle\langle\bar{\psi}| X_1 X_2 X_3 \\ \sigma'_3 &= (1-p)^2 p |\bar{\psi}\rangle\langle\bar{\psi}| + (1-p)p^2 X_1 X_2 X_3 |\bar{\psi}\rangle\langle\bar{\psi}| X_1 X_2 X_3\end{aligned}$$

The final density matrix is $\sigma' = \sum_{k=0}^3 \sigma'_k$. Does this redundancy actually help? Calculating the P_{error} , we find that the error probability has indeed reduced to p^2 from p (note $p < 1$).

$$\begin{aligned}P_{\text{error}} &= 1 - \langle\bar{\psi}|\sigma'|\bar{\psi}\rangle \\ &= p^2(3-2p) (1 - |\langle\bar{\psi}|X_1 X_2 X_3|\bar{\psi}\rangle|^2) \\ &= p^2(3-2p) (1 - |\alpha^\dagger\beta + \beta^\dagger\alpha|^2)\end{aligned}$$

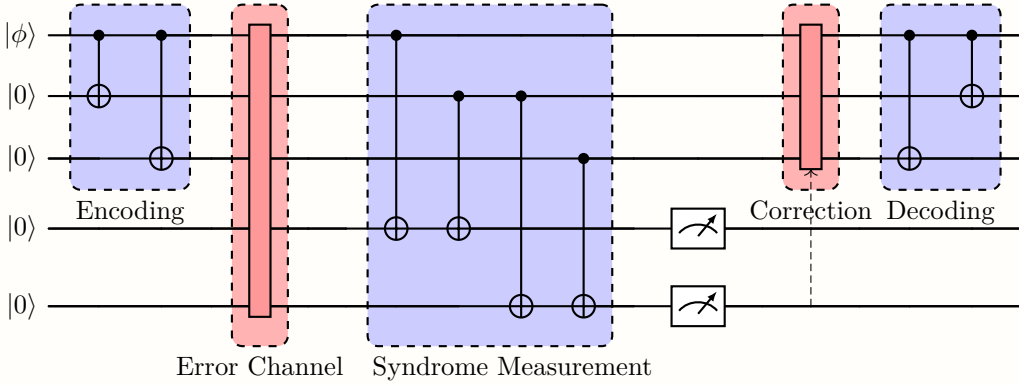


Figure 12.3: Bit Flip Error Correction Circuit

Instead of a bit flip of a single qubit, consider rotation of a single qubit around the x axis, by some angle $\theta \in \mathbb{R}$, such that the unitary operator responsible is $\mathcal{U} = e^{i\frac{\theta}{2}X} = \cos\frac{\theta}{2}\mathbb{I} + i\sin\frac{\theta}{2}X$ which can be seen as superposition of correctable errors. In this case, the error correction procedure, specifically the syndrome measurement, digitizes the superposition and discretizes the error into one of two cases. Syndrome decoding of the quantum repetition code illustrates how the original objections to the possibility of quantum error correction are overcome. It is not necessary to clone the qubit state, and the decoding procedure does not damage the superposition of the quantum information. Furthermore, the decoding procedure does discretize all errors which are superpositions of \mathbb{I} and X to a probabilistic mixture of either \mathbb{I} or X . The drawback is that this repetition code only protects against single-qubit errors of this form, and also assumes uncorrelated errors among qubits.

Multi-qubit error: If we have a multi-qubit error, our scheme of error correction might not work, as illustrated below. Suppose two physical qubits suffer bit-flips: qubits 1 and 2 are flipped, i.e. the error operator is $X_1 X_2$. The post-error state is

$$\alpha X_1 X_2 |000\rangle + \beta X_1 X_2 |111\rangle = \alpha |110\rangle + \beta |001\rangle.$$

The repetition-code syndrome is obtained from the parity checks $Z_1 Z_2$ and $Z_2 Z_3$. Evaluating their eigenvalues on the components, for $|110\rangle$: $Z_1 Z_2 |110\rangle = (+1) |110\rangle$, $Z_2 Z_3 |110\rangle = (-1) |110\rangle$; and for $|001\rangle$: $Z_1 Z_2 |001\rangle = (+1) |001\rangle$, $Z_2 Z_3 |001\rangle = (-1) |001\rangle$.

Hence the two-qubit error state $X_1 X_2 |\bar{\psi}\rangle$ has syndrome $(Z_1 Z_2, Z_2 Z_3) = (+1, -1)$. But that very same syndrome is produced by a single flip on qubit 3. Thus the syndrome measurement cannot distinguish the two situations $X_1 X_2$ versus X_3 .

In practice, the decoder maps the observed syndrome $(+1, -1)$ to the correction operator X_3 . Applying that correction to the true error $X_1 X_2$ yields $\alpha |111\rangle + \beta |000\rangle$. Instead of restoring the state, the recovery process produces a logical error. This is because parity (syndrome) measurements only detect which *pattern of parity* occurred, not the exact set of flipped qubits. Two flips in the same parity pattern can be indistinguishable from a different single flip; applying the single-flip correction then produces a logical error. This is why the three-qubit repetition code cannot correct arbitrary two-qubit errors.

Effect of bit flip on Bloch sphere: Recall that an arbitrary density matrix can be written as $\frac{1}{2}(\mathbb{I} + \vec{r} \cdot \vec{\sigma})$, where \vec{r} is the Bloch vector and $\vec{\sigma}$ is Pauli vector and $\text{Tr}(\rho^2) = \frac{1+|\vec{r}|^2}{2}$. Thus, there is a vector $\vec{r} = (r_1, r_2, r_3)$ in the Bloch sphere corresponding to every density matrix.

When ρ goes through a bit flip channel, it becomes $p\rho + (1-p)X\rho X$. As

$$\rho = \frac{\mathbb{I} + r_1 X + r_2 Y + r_3 Z}{2} \text{ and } X\rho X = \frac{\mathbb{I} + r_1 X - r_2 Y - r_3 Z}{2}$$

The Bloch vector changes after going through the bit flip channel.

$$\rho \longrightarrow p\rho + (1-p)X\rho X$$

$$(r_1, r_2, r_3) \mapsto p(r_1, r_2, r_3) + (1-p)(r_1, -r_2, -r_3)$$

The new coordinates are,

$$\begin{aligned} r'_1 &= pr_1 + (1-p)r_1 = r_1 \\ r'_2 &= pr_2 + (1-p)(-r_2) = (2p-1)r_2 \\ r'_3 &= pr_3 + (1-p)(-r_3) = (2p-1)r_3 \end{aligned}$$

Thus, the x -coordinate remains unchanged, and the y and z coordinates get squeezed by a factor of $2p-1$. This is depicted in the Fig. 12.4. As the norm of the Bloch vector $|\vec{r}|$ can only decrease in this process, the trace, $\text{Tr}(\rho^2)$, can only decrease or stay the same.

An interesting thing happens at $p = 0.5$. Both y and z coordinates vanish and the Bloch sphere becomes a projection onto the x axis.

12.2.2 Phase Flip Code

Here the error channel flips the phase of the qubit with probability p . In other words the initial density matrix $\rho = |\phi\rangle\langle\phi|$ becomes, $\rho_{\text{final}} = p\rho + (1-p)Z\rho Z$.

Can one modify the bit flip circuit to account for phase flips? Recall that $X = HZH$. So, if we apply Hadamard gates across the error channel, then any phase flip will appear like bit flips, and the same circuit used for bit flips can be used for phase flips as well.

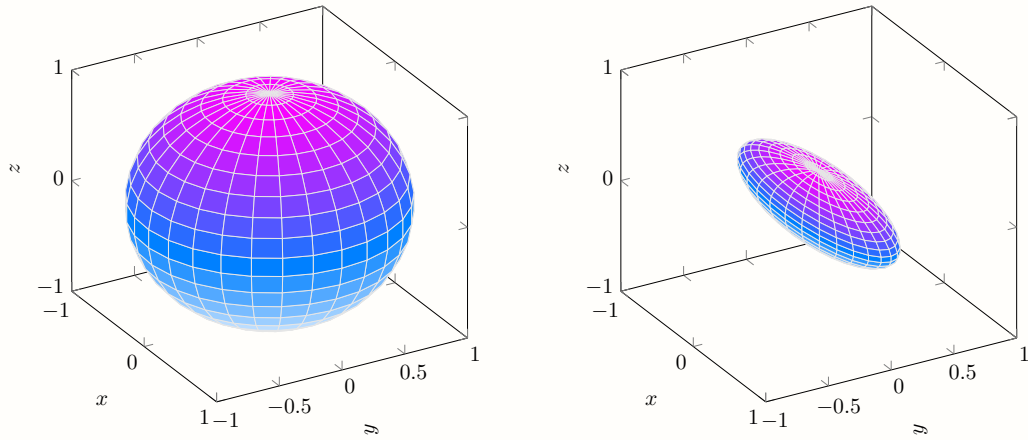
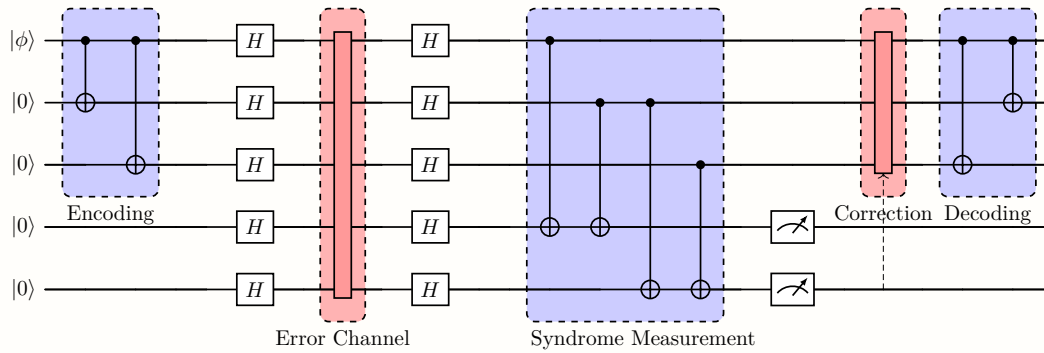
Figure 12.4: Effect of bit flip channel on the Bloch sphere, with $p = 0.2$ 

Figure 12.5: Phase Flip Error Correction Circuit

One example of encoding, correcting and decoding where the error occurs on the middle qubit is,

$$\begin{aligned} \alpha|0\rangle + \beta|1\rangle &\xrightarrow{\text{encoding}} \alpha|000\rangle + \beta|111\rangle \xrightarrow{H} \alpha|+++\rangle + \beta|---\rangle \xrightarrow{\text{error}} \alpha|+-+\rangle + \beta|-+-\rangle \\ \alpha|+-+\rangle + \beta|-+-\rangle &\xrightarrow{H} \alpha|010\rangle + \beta|101\rangle \xrightarrow{\text{correction}} \alpha|000\rangle + \beta|111\rangle \xrightarrow{\text{decoding}} \alpha|0\rangle + \beta|1\rangle \end{aligned}$$

Effect of Phase Flip Channel on the Bloch Sphere: In the case of a phase flip channel, the following is the effect on the Bloch sphere,

$$\begin{aligned} \rho &\longrightarrow p\rho + (1-p)Z\rho Z \\ (r_1, r_2, r_3) &\mapsto p(r_1, r_2, r_3) + (1-p)(-r_1, -r_2, r_3) \\ &= ((2p-1)r_1, (2p-1)r_2, r_3) \end{aligned}$$

as $ZXZ = -X$ and $ZYZ = -Y$, thus $Z\rho Z$ flips the sign of both r_1 and r_2 . Overall, the z -coordinate remains the same, and x and y coordinates get squeezed. Similar to what we saw in the case of bit flip code, at $p = 0.5$. Both x and y coordinates vanish, and the Bloch sphere becomes a projection onto the z axis.

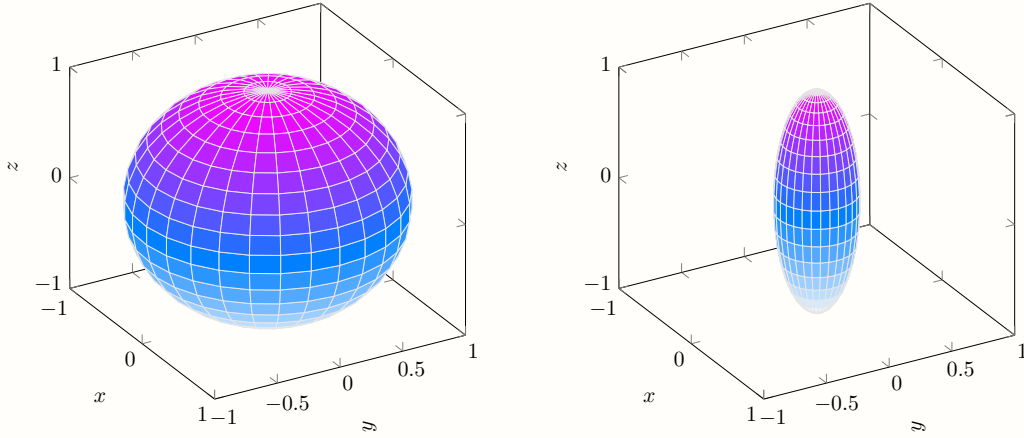


Figure 12.6: Effect of phase flip channel on the Bloch sphere, with $p = 0.2$

12.2.3 Bit-Phase Flip Code

A combination of bit and phase flip gives $XZ = -iY$. Its action on the Bloch sphere is shown in Fig. 12.7.

$$\begin{aligned} \rho &\longrightarrow p\rho + (1-p)Y\rho Y \\ (r_1, r_2, r_3) &\mapsto p(r_1, r_2, r_3) + (1-p)(-r_1, r_2, -r_3) \\ &= ((2p-1)r_1, r_2, (2p-1)r_3) \end{aligned}$$

To handle such and more general types of errors, where both bit and phase flips can occur, we need a more involved error correction code. One such code is Shor's code, which we will see in the subsequent sections.

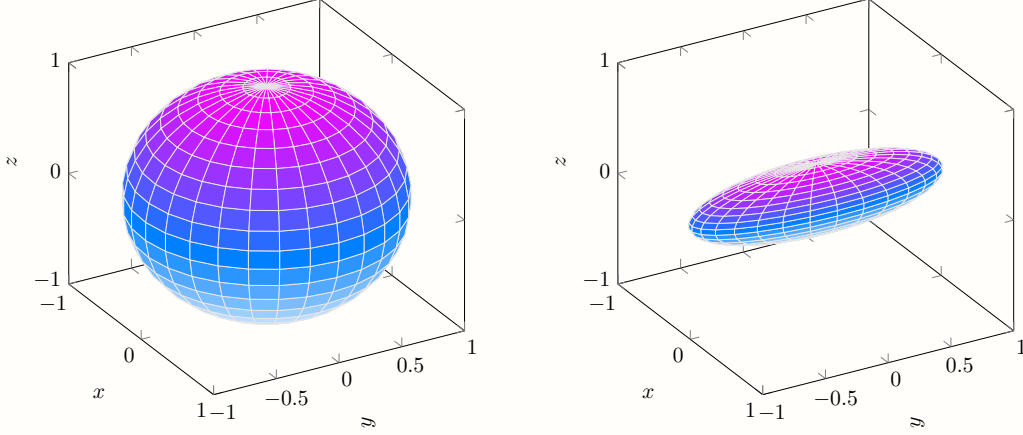


Figure 12.7: Effect of bit-phase flip (Y-flip) channel on the Bloch sphere, with $p = 0.2$

12.2.4 Depolarizing Channel

This channel represents a more general, symmetric model of noise. Intuitively, it describes a process where, with probability $1 - p$, the qubit is left untouched, and with probability p , its state is completely randomized to the maximally mixed state, $\mathbb{I}/2$. This "scrambling" effectively destroys any information stored in the qubit.

The transformation is thus described by the map:

$$\rho_{\text{final}} = (1 - p)\rho + p\frac{\mathbb{I}}{2}$$

At first glance, this transformation appears to be affine (of the form $A\rho + B$) rather than strictly linear, due to the presence of the $p\mathbb{I}/2$ term, which is independent of ρ . This might seem to conflict with the description of quantum operations as linear maps. The resolution lies from the insight that $\text{Tr}(\rho) = 1$, thereby, $\rho + X\rho X + Y\rho Y + Z\rho Z = 2\mathbb{I}$, such that,

$$\begin{aligned} \rho_{\text{final}} &= (1 - p)\rho + p \left[\frac{1}{4} (\rho + X\rho X + Y\rho Y + Z\rho Z) \right] \\ &= \left(1 - \frac{3p}{4} \right) \rho + \frac{p}{4} X\rho X + \frac{p}{4} Y\rho Y + \frac{p}{4} Z\rho Z \end{aligned}$$

confirming its linearity.

Effect of Depolarizing Channel on the Bloch Sphere: We can most easily see the geometric effect using the initial affine form. Recall $\rho = \frac{1}{2}(\mathbb{I} + \vec{r} \cdot \vec{\sigma})$.

$$\begin{aligned}\rho_{\text{final}} &= (1-p) \left[\frac{1}{2}(\mathbb{I} + \vec{r} \cdot \vec{\sigma}) \right] + p \frac{\mathbb{I}}{2} \\ &= \frac{1}{2} [\mathbb{I} + (1-p)(\vec{r} \cdot \vec{\sigma})]\end{aligned}$$

The new Bloch vector \vec{r}' is thus $\vec{r}' = (1-p)\vec{r}$.

$$(r_1, r_2, r_3) \mapsto ((1-p)r_1, (1-p)r_2, (1-p)r_3)$$

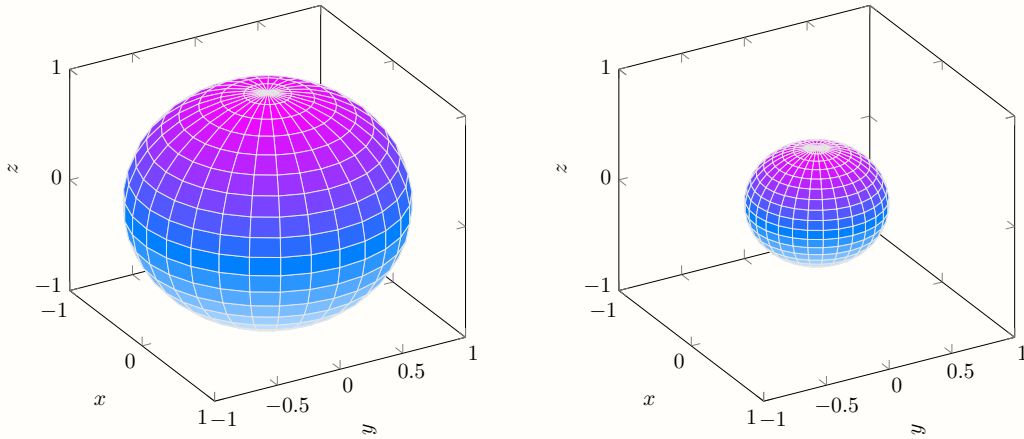


Figure 12.8: Effect of depolarizing channel on the Bloch sphere, with $p = 0.2$

Thus, the depolarizing channel uniformly shrinks the Bloch vector by a factor of $(1-p)$. The entire Bloch sphere contracts isotropically (equally in all directions) towards the center $\vec{r} = 0$, which represents the maximally mixed state $\mathbb{I}/2$. In the extreme case where $p = 1$ (complete depolarization), the new vector is $\vec{r}' = 0$, and all input states are mapped to the center, completely destroying any quantum information.

12.3 Shor's Code

We saw the bit flip and the phase flip code. Building on this, can one design an error correction scheme that can correct any arbitrary error on a single qubit? Let us try to build one such code. The critical idea is to realize that concatenating the bit flip and phase flip repetition codes produces a code which can correct any single-qubit error.

Like the earlier examples, we can start by encoding our physical qubits into a certain number of logical qubits. Suppose the state of the encoded qubit is $|\bar{\psi}\rangle = \alpha|\bar{0}\rangle + \beta|\bar{1}\rangle$, and after the action of the noise channel it becomes $\varepsilon(|\bar{\psi}\rangle\langle\bar{\psi}|) = \sum_i E_i |\bar{\psi}\rangle\langle\bar{\psi}| E_i^\dagger$. We saw

that physically we can interpret the action of the error channel as changing the initial state of the system to $E_i |\bar{\psi}\rangle \langle \bar{\psi}| E_i^\dagger$ with a certain probability. Thus, focusing on one term $E_i |\bar{\psi}\rangle \langle \bar{\psi}| E_i^\dagger$, notice that as the Pauli matrices, $\{\mathbb{I}, X, Y, Z\}$, forms a basis we can write $E_i = c_{i0}\mathbb{I} + c_{i1}X + c_{i2}XZ + c_{i3}Z$. Thus the un-normalized state $E_i |\bar{\psi}\rangle$ can be written as superposition of $\mathbb{I}|\bar{\psi}\rangle$, $X|\bar{\psi}\rangle$, $XZ|\bar{\psi}\rangle$ and $Z|\bar{\psi}\rangle$. Thus, if we have an error correction code that can correct just X and Z type errors in a single qubit, we can use it to correct E_i . As measuring the error syndrome will collapse the state to one of the above Pauli basis states, and recovery can be done appropriately.

Now let us see how to construct a code that can take care of both bit and phase flip of a single qubit. Naturally, let us just try to combine the codes we already saw, by first encoding using the phase flip code and then the bit flip code on each of the phase flip encoded qubits. This will give a 9-qubit encoding for a single qubit.

$$\begin{aligned} |0\rangle &\xrightarrow{\text{phase flip}} |+++ \rangle \xrightarrow{\text{bit flip}} \left(\frac{|000\rangle + |111\rangle}{\sqrt{2}} \right) \left(\frac{|000\rangle + |111\rangle}{\sqrt{2}} \right) \left(\frac{|000\rangle + |111\rangle}{\sqrt{2}} \right) \equiv |\bar{0}\rangle \\ |1\rangle &\xrightarrow{\text{phase flip}} |-- - \rangle \xrightarrow{\text{bit flip}} \left(\frac{|000\rangle - |111\rangle}{\sqrt{2}} \right) \left(\frac{|000\rangle - |111\rangle}{\sqrt{2}} \right) \left(\frac{|000\rangle - |111\rangle}{\sqrt{2}} \right) \equiv |\bar{1}\rangle \end{aligned}$$

The circuit to construct the above encoding is again just a combination of the bit flip and phase flip circuit as shown in Fig. 12.9.

This method of stacking and constructing an encoding is called *concatenation* and is a useful trick to construct new codes from old ones.

After encoding is done, how to correct this error? Again, we can just extend the syndrome measurements of the bit flip code and phase flip code as shown in the Table 12.2 and Table 12.3.

The complete circuit for Shor's code is given in Fig. 12.10. Note that, unlike bit flip and phase flip code, this circuit does not have a measurement operator, as measurements can be equivalently converted to control not gates 5.5.

It is important to note what type of errors Shor's code can correct and cannot. If we cluster each of the nine qubits into sets of three, then Shor's code can correct:

- Single qubit bit flip error in any 1 out of 3 qubits in a cluster.
- Phase flip in 1 cluster.

The combination of the above two covers all types of single-qubit errors. However, Shor's code cannot correct for:

- Two bit flips in a single cluster.

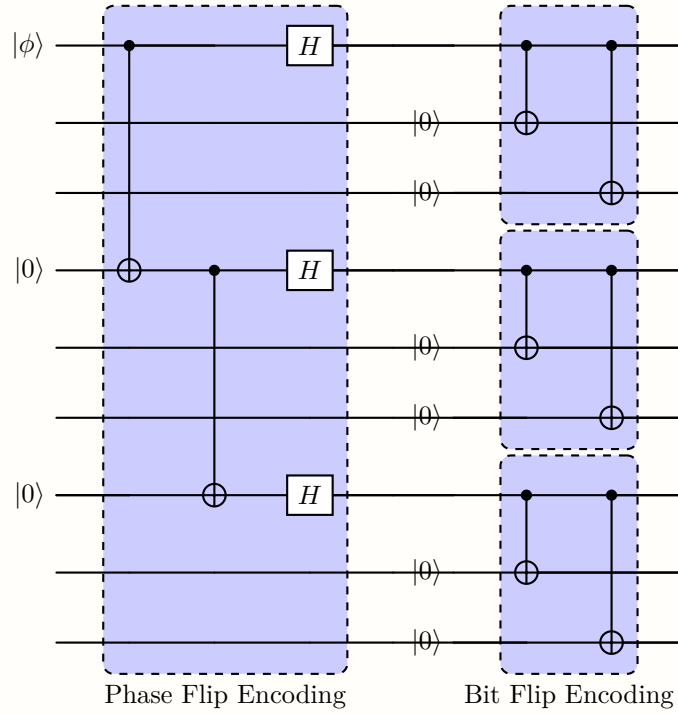


Figure 12.9: Concatenation of bit flip and phase flip encoding

Error	Z_1Z_2	Z_2Z_3	Z_4Z_5	Z_5Z_6	Z_7Z_8	Z_8Z_9	Action to correct the error
\mathbb{I}	+1	+1	+1	+1	+1	+1	\mathbb{I}
X_1	-1	+1	+1	+1	+1	+1	X_1
X_2	-1	-1	+1	+1	+1	+1	X_2
X_3	+1	-1	+1	+1	+1	+1	X_3
X_4	+1	+1	-1	+1	+1	+1	X_4
X_5	+1	+1	-1	-1	+1	+1	X_5
X_6	+1	+1	+1	-1	+1	+1	X_6
X_7	+1	+1	+1	+1	-1	+1	X_7
X_8	+1	+1	+1	+1	-1	-1	X_8
X_9	+1	+1	+1	+1	+1	-1	X_9

Table 12.2: Syndrome measurements to detect single bit flip errors. (Note: Z_4Z_5 , Z_5Z_6 , etc. are shown for clarity, representing the 6 stabilizers for the 3 bit-flip blocks).

Error	$X_1 X_2 X_3 X_4 X_5 X_6$	$X_4 X_5 X_6 X_7 X_8 X_9$	Action
\mathbb{I}	+1	+1	\mathbb{I}
$Z_1/Z_2/Z_3$	-1	+1	$Z_1/Z_2/Z_3$
$Z_4/Z_5/Z_6$	-1	-1	$Z_4/Z_5/Z_6$
$Z_7/Z_8/Z_9$	+1	-1	$Z_7/Z_8/Z_9$

Table 12.3: Syndrome measurements to detect single phase flip errors

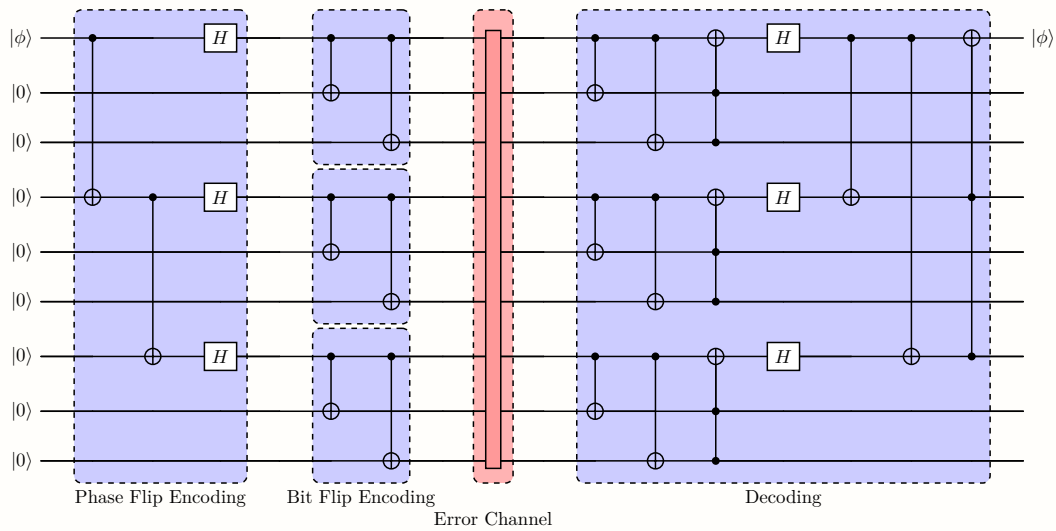


Figure 12.10: Shor's Error Correction Code

Suppose our logical state is $|\bar{\psi}\rangle = \alpha|\bar{0}\rangle + \beta|\bar{1}\rangle$, and the system suffers a two-qubit bit-flip error $E = X_1X_2$ in the first cluster (qubits 1-3).

The state becomes $E|\bar{\psi}\rangle = \alpha(X_1X_2|\bar{0}\rangle) + \beta(X_1X_2|\bar{1}\rangle)$.

The decoder checks the bit-flip syndromes for the first block: Z_1Z_2 and Z_2Z_3 . The error state in this block is $\propto X_1X_2(|000\rangle \pm |111\rangle) = (|110\rangle \pm |001\rangle)$. Measuring the stabilizers on this error state, both components give the same syndrome: $(+1, -1)$.

The syndrome $(+1, -1)$ for the first block corresponds to a *single* bit-flip on qubit 3. The decoder, assuming a single error is most probable, applies the recovery operation $R = X_3$.

The resulting operator $F = X_1X_2X_3$ is not the identity. It is, in fact, the logical Z operator for the first block, \bar{z}_1 .

This operator commutes with all the bit-flip stabilizers (e.g., $(X_1X_2X_3)(Z_1Z_2) = (X_1Z_1)(X_2Z_2)X_3 = (-Z_1X_1)(-Z_2X_2)X_3 = Z_1Z_2X_1X_2X_3$) and is thus undetectable by them. It also commutes with the phase-flip stabilizers.

Applying this correction has finalized a logical error on the first block, corrupting the encoded state $|\bar{\psi}\rangle$ instead of restoring it.

- Phase error in two different clusters.

Suppose the system suffers a phase-flip error on qubit 1 (first cluster) and qubit 4 (second cluster).

The error operator is $E = Z_1Z_4$. The state is $E|\bar{\psi}\rangle = (Z_1Z_4)|\bar{\psi}\rangle$, with decoder operators $M_1 = X_1X_2X_3X_4X_5X_6$ and $M_2 = X_4X_5X_6X_7X_8X_9$.

For M_1 , The error $E = Z_1Z_4$ anticommutes with X_1 and X_4 . Since it anticommutes with an even number (two) of the X operators in M_1 , it *commutes* with M_1 overall. $M_1E = M_1(Z_1Z_4) = (X_1Z_1)(X_4Z_4)\cdots = (-Z_1X_1)(-Z_4X_4)\cdots = Z_1Z_4M_1 = EM_1$. The syndrome is $+1$. For M_2 , The error $E = Z_1Z_4$ anticommutes with X_4 but commutes with all other operators in M_2 . Since it anticommutes with an odd number (one) of the X operators, it *anticommutes* with M_2 overall. $M_2E = M_2(Z_1Z_4) = Z_1(X_4Z_4)\cdots = Z_1(-Z_4X_4)\cdots = -EM_2$. The syndrome is -1 .

The measured syndrome is $(+1, -1)$ and identifies this syndrome as corresponding to a phase error in the *third* cluster (e.g., Z_7 , Z_8 , or Z_9). The decoder applies the recovery operation $R = Z_7$. The total operator applied to the state is $F = R \cdot E = Z_7(Z_1Z_4) = Z_1Z_4Z_7$.

The operator $F = Z_1Z_4Z_7$ is a logical bit-flip, \bar{X} , for the Shor code. The correction has combined with the two-qubit error to produce a logical \bar{X} operation, $E|\bar{\psi}\rangle \rightarrow F|\bar{\psi}\rangle = \bar{X}|\bar{\psi}\rangle$. This is a catastrophic failure of the code, as the encoded information has been flipped.

The keen reader would have wondered and probably noticed a deep connection running through our examples. In the bit-flip code, our logical states $|\bar{0}\rangle = |000\rangle$ and $|\bar{1}\rangle = |111\rangle$ were precisely the states that were *stabilized*, with eigenvalue $+1$; by the syndrome measurement operators Z_1Z_2 and Z_2Z_3 . The same principle held for the phase-flip code and its X_1X_2 , X_2X_3 operators, and indeed for the more complex Shor code.

This observation is far from a coincidence; it is the very essence of the code. The protected subspace $\text{Span}\{|0\rangle, |1\rangle\}$ is defined as the simultaneous $+1$ eigenspace of these operators. This raises a powerful question: Instead of constructing codes by hand and then finding their syndrome operators, could we reverse the process? What if we start by choosing a set of commuting operators (drawn from the Pauli group, naturally) and simply define our code space as the subspace they all stabilize?

This is precisely the idea behind the stabilizer formalism. It provides an elegant and powerful algebraic framework for describing, constructing, and analyzing quantum error-correcting codes.

12.4 Formalisms

We rigorously introduce the Kraus operator formalism to describe the error channels and understand their correctability. Further, since it is a lot more convenient to describe the Shor code and many other quantum error-correcting codes in terms of their stabilizers, we shall delve into the depths of the formalism.

12.4.1 Kraus Operator in Error Correction

The concept of Kraus operator was introduced in the Sec. 9.2. For an interacting system with initial independent density matrix ρ , we have the *operator-sum representation*,

$$\rho' = \sum_k E_k \rho E_k^\dagger$$

where E_k is the Kraus operator on the principal system Hilbert space. Physically, the action of quantum operation is equivalent to taking the system from ρ and randomly placing it in normalized $\frac{1}{\text{Tr}(E_k \rho E_k^\dagger)} E_k \rho E_k^\dagger$ with probability $\text{Tr}(E_k \rho E_k^\dagger)$, which is similar to the effect of a noisy classical communication channel.

The above shows that given an interacting quantum system, it gives rise to an operator-sum representation. But is the converse true? Given a set of operators $\{E_k\}$, can we associate it with a model environmental system and dynamics that gives the corresponding operator-sum representation, where the dynamics is either unitary or projective measurements? The answer turns out to be yes!¹

We shall now see a few examples of errors and a method for correcting them. Given the above operator language, one can now associate Kraus operators (as shown in Table 12.4) with all the error correction codes seen so far.

Revisiting the earlier examples, the Kraus Operators for the relevant codes are given in Table 12.4.

Quantum Error Correction Condition in Operator Formalism A natural question to ask is, can any form of error be corrected? Here comes the power of operator representation, which can characterize correctable errors.

¹Proof for this can be found in the Nielsen, M.A. and Chuang, I.L. [2011].

Error Type	Kraus Operator	Syndrome	Bloch Sphere: $(r_1, r_2, r_3) \rightarrow$
Bit Flip	$\{\sqrt{1-p}\mathbb{I}, \sqrt{p}X\}$	Z_1Z_2 and Z_2Z_3	$(r_1, (1-2p)r_2, (1-2p)r_3)$
Phase Flip	$\{\sqrt{1-p}\mathbb{I}, \sqrt{p}Z\}$	X_1X_2 and X_2X_3	$((1-2p)r_1, (1-2p)r_2, r_3)$
Bit-Phase Flip	$\{\sqrt{1-p}\mathbb{I}, \sqrt{p}Y\}$	Z_1Z_2 and Z_2Z_3	$((1-2p)r_1, r_2, (1-2p)r_3)$

Table 12.4: Summary of the error codes with corresponding Kraus operators. (Note: The Bloch sphere transformations assume the standard channel $\rho' = (1-p)\rho + pE\rho E^\dagger$, which may differ from the convention used in Sec 2.)

Theorem 12.4.1. *Let C be a quantum code and Π projector onto C . Suppose ε (noise) is a quantum operator with operation elements $\{E_i\}$, then the necessary and sufficient condition for the existence of an error-correcting operation R correcting ε on C is that*

$$\Pi E_i^\dagger E_j \Pi = m_{ij} \Pi$$

for some Hermitian matrix $\mathbf{m} = [m_{ij}]$. The above condition is called quantum error correction condition or the Knill-Laflamme (KL) condition. The operator elements $\{E_i\}$ for the noise ε are called errors, and if such R exists, we say $\{E_i\}$ constitutes a correctable set of errors.

Proof. We will prove the sufficiency by constructing R given the KL condition.

(\Rightarrow): Suppose $\{E_i\}$ satisfies KL condition, $\mathbf{m} = [m_{ij}]$ is Hermitian so it can be diagonalized $\Lambda = U^\dagger \mathbf{m} U$. Define $F_k \equiv \sum_i U_{ik} E_i$. As F_k is a sum of unitaries times operator elements, we know that $\{F_k\}$ is also a set of operator elements of ε .

Substituting F_k in KL condition we get,

$$\begin{aligned} \Pi F_k^\dagger F_l \Pi &= \sum_{ij} U_{ki}^\dagger U_{jl} \Pi E_i^\dagger E_j \Pi \\ &= \sum_{ij} U_{ki}^\dagger U_{jl} m_{ij} \Pi = \sum_{ij} U_{ki}^\dagger m_{ij} U_{jl} \Pi \\ &= \Lambda_{kl} \Pi \end{aligned}$$

Since $F_k \Pi$ is a linear map, we can find a polar decomposition

$$F_k \Pi = \mathcal{U}_k \sqrt{\Pi F_k^\dagger F_k \Pi} = \sqrt{\Lambda_{kk}} \mathcal{U}_k \Pi$$

since $\Pi^2 = \Pi \Rightarrow \sqrt{\Pi} = \Pi$, and for some unitary \mathcal{U}_k .

We have $F_k \Pi = \sqrt{\Lambda_{kk}} \mathcal{U}_k \Pi$, implying, $\mathcal{U}_k \Pi \mathcal{U}_k^\dagger = \frac{F_k \Pi \mathcal{U}_k^\dagger}{\sqrt{\Lambda_{kk}}}$. Now, define $\Pi_k \equiv U_k \Pi U_k^\dagger$, then,

$$\Pi_l \Pi_k = \Pi_l^\dagger \Pi_k = (\mathcal{U}_l^\dagger \Pi \mathcal{U}_l)(\mathcal{U}_k \Pi \mathcal{U}_k^\dagger) = \frac{\mathcal{U}_l^\dagger \Pi F_l^\dagger}{\sqrt{\Lambda_{ll}}} \frac{F_k \Pi \mathcal{U}_k^\dagger}{\sqrt{\Lambda_{kk}}} = \frac{\mathcal{U}_l^\dagger}{\sqrt{\Lambda_{ll}}} \Pi F_l^\dagger F_k \Pi \frac{\mathcal{U}_k^\dagger}{\sqrt{\Lambda_{kk}}} = 0$$

Note that $\Pi F_l^\dagger F_k \Pi = \Lambda_{lk} \Pi$ and it is 0 when $l \neq k$. Thus, Π_k 's can be thought of as orthonormal projective measurements.

Now the syndrome measurement can be defined as the projectors Π_k augmented by an additional projector, if necessary, to satisfy the completeness relation. Let $\Pi' = \mathbb{I} - \sum_k \Pi_k$. Therefore, $\{F_k\}$ being equivalent to $\{E_k\}$, has the action on the code space, as

$$F_k \Pi = \sqrt{\Lambda_{kk}} \mathcal{U}_k \Pi$$

Note that we can get back to the code space by applying \mathcal{U}_k^\dagger .

Thus, the combined detection-recovery step corresponds to the quantum operator

$$R(\sigma) = \sum_k \mathcal{U}_k^\dagger \Pi_k \sigma \Pi_k \mathcal{U}_k$$

Note that, we have

$$\begin{aligned} \mathcal{U}_k^\dagger \Pi_k F_l \sqrt{\rho} &= \mathcal{U}_k^\dagger \Pi_k^\dagger F_l \Pi \sqrt{\rho} = \frac{\mathcal{U}_k^\dagger \mathcal{U}_k \Pi F_k^\dagger F_l \Pi}{\sqrt{\Lambda_{kk}}} \sqrt{\rho} \\ &= \delta_{kl} \sqrt{\Lambda_{kk}} \Pi \sqrt{\rho} \end{aligned}$$

Therefore, we have

$$R(\varepsilon(\rho)) = \sum_{kl} \mathcal{U}_k^\dagger \Pi_k F_l \rho F_l^\dagger \Pi_k \mathcal{U}_k = \sum_{kl} \delta_{kl} \Lambda_{kk} \rho \propto \rho$$

as required by the recovery operation.

(\Leftarrow): Suppose $\{E_i\}$ is a set of errors which is perfectly correctable by an error-correction operation R with operation elements $\{R_j\}$. Define the quantum operator $\varepsilon_c(\rho) \equiv \varepsilon(\Pi \rho \Pi)$ where $\Pi \rho \Pi$ is in the code space.

We thereby have, $R(\varepsilon_c(\rho)) \propto \Pi \rho \Pi$, expanding this out $\sum_{ij} R_j E_i \Pi \rho \Pi E_i^\dagger R_j^\dagger = \mathcal{C} \Pi \rho \Pi$, where \mathcal{C} is constant.

Thereby, $\{R_j E_i\}$ is identical to a single operator element $\sqrt{\mathcal{C}} \Pi$, such that $R_k E_i \Pi = \mathcal{C}_{ki} \Pi$, taking adjoint $\Pi E_i^\dagger R_k^\dagger = \mathcal{C}_{k,i}^\dagger \Pi$, therefore $\Pi E_i^\dagger R_k^\dagger R_k E_j \Pi = \mathcal{C}_{k,i}^\dagger \mathcal{C}_{k,j} \Pi$

As R is trace-preserving $\sum_k R_k^\dagger R_k = 1$, therefore,

$$\sum_k \Pi E_i^\dagger R_k^\dagger R_k E_j \Pi = \Pi E_i^\dagger E_j \Pi = \sum_k \mathcal{C}_{k,i}^\dagger \mathcal{C}_{k,j} \Pi = m_{ij} \Pi$$

where m_{ij} is Hermitian. This matches with the KL condition. ■

Applying the KL criterion to the Shor code

For Shor's code, we have the correctable error set, described by $E = \{\mathbb{I}, X_i, Y_i, Z_i\}$ for $i = 1, 2, \dots, 9$. We choose the code basis as $|\bar{0}\rangle$ and $|\bar{1}\rangle$. It is easy to note that

$\langle \bar{0} | E_i^\dagger E_j | \bar{1} \rangle = 0$, since the basis kets are constructed orthogonally, and there is no transformation connecting the two.

The significant condition to check is for whether $\langle \bar{0} | E_i^\dagger E_j | \bar{0} \rangle = \langle \bar{1} | E_i^\dagger E_j | \bar{1} \rangle$, where usually both are not zero. For $E_i = Z_\alpha$, $E_j = Z_\beta$, we have both equal to one, and similarly for other combinations of Z_α 's. For any other operator, both quantities are equal to zero. Thereby, the KL condition is satisfied for Shor's.

12.4.2 Stabilizer Formalism for Error Correction

The stabilizer formalism offers a streamlined approach to identifying and correcting errors.

12.4.2.1 Pauli Group

As seen before, the four Pauli operators, \mathbb{I}, X, Z, Y allow us to express the four possible effects of the environment on a qubit. These operators form a group $G = \mathcal{P}$ and they exhibit several nice properties:

- Anticommuting

$$\{X, Z\} = \{Y, Z\} = \{X, Y\} = 0$$

- $P^2 = \mathbb{I}$, for all $P \in \mathcal{P}$
- Span the space of 2×2 matrices, describing the transformation of a single qubit.

Further, two Pauli matrices are equivalent if $\sigma_j = c\sigma_i$, where $c = \pm 1, \pm i$, which lets us define the set of equivalence classes of Pauli operators $[\mathcal{P}]$. Note that the set of Pauli operators, \mathcal{P} , is not an Abelian group. However, the set Π of equivalence classes, $[\mathcal{P}]$, of Pauli operators, also called the projective Pauli group, forms an Abelian group.

We further define the 1-qubit Pauli group, \mathcal{P}_1 , which consists of the Pauli operators, \mathbb{I}, X, Y, Z , together with the multiplicative factors, $\pm 1, \pm i$, as

$$\mathcal{P}_1 := \{\pm \mathbb{I}, \pm i\mathbb{I}, \pm X, \pm iX, \pm Z, \pm iZ, \pm Y, \pm iY\}$$

whose cardinality is $|\mathcal{P}_1| = 16$. The members of the 1-qubit Pauli group are unitary, either commute or anticommute, and are either Hermitian or anti-Hermitian. Note that the generators of \mathcal{P}_1 are $\mathcal{G}_1 = \{X, Z, i\mathbb{I}\}$.

Generalizing our ideas, the n -qubit *Pauli group* \mathcal{P}_n consists of the 4^n tensor products \mathbb{I}, X, Y, Z and an overall phase of $\pm 1, \pm i$, thereby, the group has $4 \times 4^n = 4^{n+1}$ elements (can be verified for $|\mathcal{P}_1| = 4^{1+1} = 16$). One element of the group is an n -tuple, the tensor product of n one-qubit Pauli operators, and can be used to describe the error operator applied to an n -qubit register. We define the *weight* of such an operator in \mathcal{P}_n to be the number of tensor factors that are not equal to \mathbb{I} .

12.4.2.2 Stabilizer Subgroup

The stabilizer formalism is a concise way to describe a quantum error-correcting code using a set of quantum operators. Assume that the m codewords of a code are represented by the vectors $|\psi\rangle$, which are n -qubit registers. We have identified a set of q operators M_j that enable us to detect errors that may affect any of the codewords. In this context, the term *detect* refers to a measurement process that does not disclose any information about the actual state, but rather indicates whether the codeword has been affected by errors.

The stabilizer \mathcal{S} of a quantum code is an Abelian subgroup of the n -qubit Pauli group, with generators $\{M_1, M_2, \dots\}$ where M_i stabilises the code words with positive eigenvalue. The code space C is the simultaneous $+1$ eigenspace of all generators. The codewords are thereby the eigenvectors satisfying $M_j|\psi\rangle = (+1)|\psi\rangle$ for all j and all $|\psi\rangle \in C$. For an error $|\varphi\rangle = E_\alpha|\psi\rangle$ due to the error operator E_α , the stabilizers act as *syndrome* measurements. If E_α anticommutes with M_j , then $M_j|\varphi\rangle = (-1)|\varphi\rangle$, revealing a non-trivial syndrome.

The *normalizer* $\mathcal{N}(\mathcal{S})$ is the set of elements that fix the stabilizer code under conjugation. Further, $\mathcal{S} \subset \mathcal{N}(\mathcal{S})$ is a normal subgroup. Since the elements of the normalizer $\mathcal{N}(\mathcal{S})$ move codewords around in the code space, they are the *logical* operators. Only the elements $E \in \mathcal{N}(\mathcal{S}) - \mathcal{S}$ act on the codewords nontrivially.

The *centralizer* $\mathcal{C}(\mathcal{S})$ is the set of elements in \mathcal{P}_n that commute with all the elements of the stabilizer \mathcal{S} . Since \mathcal{S} can be shown to be an Abelian subgroup, it can be shown that the normalizer equals the centralizer, $\mathcal{N}(\mathcal{S}) = \mathcal{C}(\mathcal{S})$.

12.4.2.3 Quantum Error Correction Condition in stabilizer Formalism

Consider $|\psi\rangle$ as a codeword in the code space C , with a set of stabilizer generators $M_j \in \mathcal{S}$, such that

$$M_j|\psi\rangle = (+1)|\psi\rangle \quad \text{for all } M_j \in \mathcal{S} \text{ and } |\psi\rangle \in C$$

The errors, $E = \{E_1, E_2, \dots\}$, affecting a codeword are also a subgroup of the n -qubit Pauli group, $E \in \mathcal{P}_n$, with each error operator E_i being a tensor product of n Pauli matrices. The *weight* of an error operator is equal to the number of errors affecting a quantum word, thus, the number of Pauli operators other than \mathbb{I} in this n -dimensional tensor product. Note that the correctable error operators anticommute with at least one of the generators of the stabilizer group \mathcal{S} . That is, for a given E , there exists some M_j such that

$$M_j(E|\psi\rangle) = (-1)E(M_j|\psi\rangle) = (-1)E|\psi\rangle$$

since $\{M_j, E\} = 0$. Therefore, to detect errors, we have to compute the eigenvalues of the generators and identify those with an eigenvalue of -1 .

For the stabilizer \mathcal{S} , with $n - k$ generators, then it encodes k qubits, and code distance d , which accounts for the smallest number of qubits that can be in error such that the error is undetectable by the code². We thereby, compactify, as a $[n, k, d]$ stabilizer code, with n being the length of a codeword (number of physical qubits), k the number of information

²We assume that the identity operator is a correctable error.

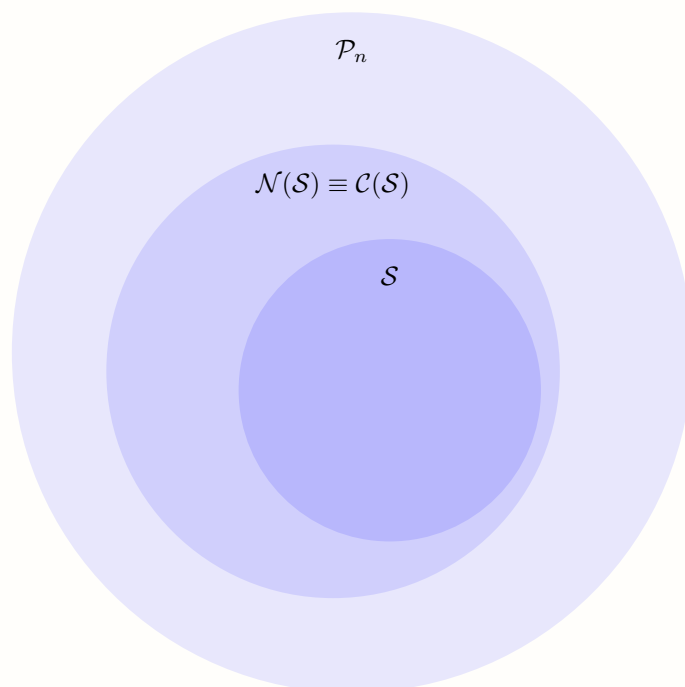


Figure 12.11: Error pattern classification for stabilizer codes. Correctable Errors: $E \in \mathcal{P}_n - \mathcal{N}(\mathcal{S}) \equiv \mathcal{P}_n - \mathcal{C}(\mathcal{S})$. Non Detectable Errors: $E \in \mathcal{C}(\mathcal{S}) - \mathcal{S}$.

symbols (logical qubits), and d being the distance of the code, such that the cardinality of the stabilizer is $|\mathcal{S}| = 2^{n-k}$. Rather than specifying the entire group, we only need its $n - k$ independent generators. Each generator imposes a constraint that effectively halves the dimension of the available Hilbert space. Thus, starting with n physical qubits (a 2^n dimensional space), these $n - k$ constraints define a 2^{n-k} times smaller subspace of dimension 2^k , perfectly suited to encode k logical qubits.

The code's power is its distance d , which measures its resilience. An error E is detected if it anti-commutes with a stabilizer, producing a measurable syndrome. An error is undetectable if it commutes with all stabilizers. The set of all such commuting operators is the normalizer, $\mathcal{N}(\mathcal{S})$.

Undetectable errors fall into two categories: *Trivial Errors*: If the error E is an element of the stabilizer itself ($E \in \mathcal{S}$), it is harmless as it leaves the code states unchanged; *Logical Errors*: If the error E is in the normalizer but not the stabilizer ($E \in \mathcal{N}(\mathcal{S}) - \mathcal{S}$), it acts as a non-trivial operation on the encoded logical qubits (e.g., a logical bit-flip \bar{X}). This corrupts the information without being flagged.

The code distance d is therefore the minimum weight (the number of qubits it acts on non-trivially) of an operator in this set of dangerous logical errors, $\mathcal{N}(\mathcal{S}) - \mathcal{S}$. This is the smallest error that can silently damage the encoded data. In terms of the KL conditions, $\Pi E \Pi \neq C \Pi$ for some c for an operator $E \in \mathcal{P}_n$ of weight d . Thus, a code of distance d can correct all Pauli errors of weight no larger than $\lfloor \frac{d-1}{2} \rfloor$, since the projection of any two such errors has weight twice that of it.

12.4.3 Quantum Hamming Bound

A natural question to ask is how efficiently we can make the error correction code in terms of the number of physical qubits used. Can we have an arbitrarily small number of physical qubits? If not, what is the smallest number of physical qubits needed?

To answer the above questions, let's suppose a non-degenerate code is used to encode k -qubits in n -qubits such that it can correct errors on any subset of t or fewer qubits. Suppose $j \leq t$ errors occur, then the number of possible locations where this can happen is $\binom{n}{j}$. If in each of these locations, the errors can be one of X, Y , or Z , then the total number of errors that may occur on t or fewer qubits is $\sum_{j=0}^t \binom{n}{j} 3^j$.

As the code is non-degenerate, to encode k qubits we need a 2^k -dimensional space. Thus, each of the above errors must correspond to an orthogonal 2^k -dimensional subspace. As we are encoding with n qubits, this number should be less than 2^n , the dimension of the physical qubit states. Therefore, we have,

$$\sum_{j=0}^t \binom{n}{j} 3^j 2^k \leq 2^n$$

The above condition is known as the *quantum Hamming bound*.

For correcting a single qubit error, we have $k = 1$ and $t = 1$. On substituting this, the quantum Hamming bound gives us $(1 + 3n)2^1 \leq 2^n$, or $1 + 3n \leq 2^{n-1}$. This inequality is first satisfied for $n = 5$. Thus, to answer the question asked at the start of this section, to correct an arbitrary single-qubit error, we need at least 5 physical qubits. Do we have an error correction code that works on exactly 5 physical qubits? The answer turns out to be yes!

The Perfect Code

The five-qubit code encodes $k = 1$ logical qubit into $n = 5$ physical qubits and has a distance $d = 3$, allowing it to correct $t = \lfloor (3-1)/2 \rfloor = 1$ arbitrary single-qubit error. Its stabilizer \mathcal{S} is defined by $n - k = 4$ generators

$$M_1 = X_1 Z_2 Z_3 X_4 \mathbb{I}_5$$

$$M_2 = \mathbb{I}_1 X_2 Z_3 Z_4 X_5$$

$$M_3 = X_1 \mathbb{I}_2 X_3 Z_4 Z_5$$

$$M_4 = Z_1 X_2 \mathbb{I}_3 X_4 Z_5$$

The corresponding logical operators that act on the protected code space C are $\bar{X} = X_1 X_2 X_3 X_4 X_5$ and $\bar{Z} = Z_1 Z_2 Z_3 Z_4 Z_5$. \bar{X} and \bar{Z} both commute with all stabilizer generators (e.g., \bar{X} anticommutes twice with M_1) and that they anticommute with each other, as required for a logical qubit.

A Not-So-Perfect Code

The 5-qubit code is special because it's *perfect*. The 7 qubit Steane code also encodes $k = 1$ logical qubit and corrects $t = 1$ single-qubit error with code distance $d = 3$. Unlike the 5-qubit code, the Quantum Hamming Bound is *not* saturated.

The Steane code's stabilizer \mathcal{S} is defined by $n - k = 6$ generators. They are grouped into X -type and Z -type, which allows for separate correction of bit-flips and phase-flips:

$$M_1 = Z_1 Z_3 Z_5 Z_7$$

$$M_4 = X_1 X_3 X_5 X_7$$

$$M_2 = Z_2 Z_3 Z_6 Z_7$$

$$M_5 = X_2 X_3 X_6 X_7$$

$$M_3 = Z_4 Z_5 Z_6 Z_7$$

$$M_6 = X_4 X_5 X_6 X_7$$

The logical operators are also beautifully symmetric

$$\bar{X} = X_1 X_2 X_3 X_4 X_5 X_6 X_7 \quad \bar{Z} = Z_1 Z_2 Z_3 Z_4 Z_5 Z_6 Z_7$$

Measuring the Z -type stabilizers provides the syndrome for X errors, and measuring the X -type stabilizers provides the syndrome for Z errors.

12.5 Surface Codes

We now move to a particularly important and promising class of stabilizer codes: *surface codes*. These codes get their name from their layout on a two-dimensional surface or lattice, where qubits are typically placed on the edges (or vertices/faces) and stabilizer generators are defined as local operators acting on small, neighboring sets of qubits.

The essential idea is to encode logical information *non-locally* in the global topological properties of this surface. This topological protection is what grants them their resistance to local errors and their high error threshold, making them a leading candidate for building fault-tolerant quantum computers.

12.5.1 Toric Code

Imagine a square lattice, but with its end edges identified and seen as a torus. This is extremely similar to periodic boundary conditions in other aspects of physics, where we exploit the periodicity of the square lattice with endpoints identified. We can imagine assigning operators along the individual points of the lattice. Fig. 12.12 represents the toric code setup, where solid lines gives the lattice, and on each edge of the lattice lies a blue dot which represents a qubit. For an $n \times n$ lattice, we have $2n^2$ qubits since we can imagine each square hosting $\frac{1}{2} \times 4$ qubits, summing on each of its edges.

We define two types of stabilizer generators on the lattice as, Star operators \mathcal{Q}_s and Plaquette operators \mathcal{B}_p as

$$\mathcal{Q}_s = \prod_{j \in \text{Star}(s)} X_j, \quad \mathcal{B}_p = \prod_{j \in \text{Plaquette}(p)} Z_j,$$

where we note that \mathcal{Q}_s and \mathcal{B}_p individually commute, and also commute with each other for any pair of s, p . This can be explicitly seen through noting

$$\begin{aligned} [\mathcal{Q}_s, \mathcal{Q}_{s'}] &= \left[\prod_{j \in \text{Star}(s)} X_j, \prod_{k \in \text{Star}(s')} X_k \right] \\ &= \prod_{j \in \text{Star}(s), k \in \text{Star}(s')} [X_j, X_k] = 0, \end{aligned}$$

since the individual X operators are independent on different locations s and s' given by X_j and X_k respectively, and the commutator product simplifies. Similarly,

$$\begin{aligned} [\mathcal{B}_p, \mathcal{B}_{p'}] &= \left[\prod_{j \in \text{Plaquette}(p)} Z_j, \prod_{k \in \text{Plaquette}(p')} Z_k \right] \\ &= \prod_{j \in \text{Plaquette}(p), k \in \text{Plaquette}(p')} [Z_j, Z_k] = 0. \end{aligned}$$

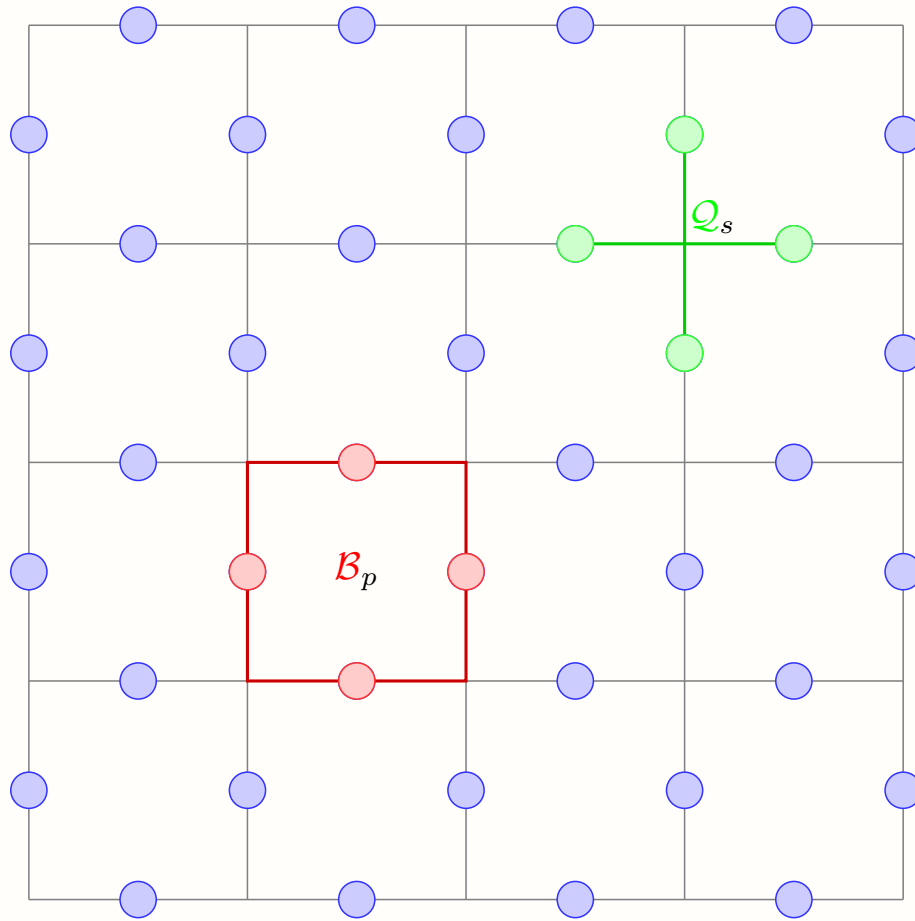


Figure 12.12: Toric Code

Further, we note the interesting relation,

$$\begin{aligned} [\mathcal{Q}_s, \mathcal{B}_p] &= \left[\prod_{j \in \text{Star}(s)} X_j, \prod_{k \in \text{Plaquette}(p)} Z_k \right] \\ &= \prod_{j \in \text{Star}(s), k \in \text{Plaquette}(p)} [X_j, Z_k] \end{aligned}$$

where we analyse the cases. When X_j and Z_k are widely separated as in Fig. 12.12, we note the independency and thereby, $[X_j, Z_k] = 0$. But, when we have overlapping cases for X_j and Z_k as in Fig. 12.13. Here, note that there always exist two overlaps, say i_1 and i_2 . Hence, we have the product $[\mathcal{Q}_s, \mathcal{B}_p] \sim [\dots X_{i_1} X_{i_2} \dots, \dots Z_{i_1} Z_{i_2} \dots]$, where these overlaps cancel each other out through their commutation relations. Since the overlap between the star and plaquette operator always ends up in overlap on even number of qubits, it is easy to show that $[\mathcal{Q}_s, \mathcal{B}_p] = 0$.

Thereby, it is interesting to note that there are n^2 Star operators defined at each \mathcal{Q}_s for the X operator, and similarly, n^2 Plaquette operators at each \mathcal{B}_p for the Z operator. But these aren't all independent and are connected by a simple relation. Note that, in the product of all Star and Plaquette operators, we encounter the X and Z respectively twice since adjacent stars and neighbouring Plaquettes share a common qubit. This results in products of $X^2 = \mathbb{I}$ and $Z^2 = \mathbb{I}$, across all qubits, resulting in

$$\prod_s \mathcal{Q}_s = \prod_s \left(\prod_{j \in \text{Star}(s)} X_j \right) = \mathbb{I}, \quad \prod_p \mathcal{B}_p = \prod_p \left(\prod_{j \in \text{Plaquette}(p)} Z_j \right) = \mathbb{I}.$$

Thereby, we have one of the Star and Plaquette operators being determined through the others, leading to $2(n^2 - 1) = 2n^2 - 2$ independent stabilizer operators in total. Thereby, we encode in dimension $2^{2n^2 - (2n^2 - 2)} = 2^2$, that is, we encode 2 qubits into $2n^2$ qubits.

We further define the set of logical operators as cycles on the torus, as in Fig. 12.14, defined as cycles since the edges are identified. Precisely, we have

$$\mathcal{X}_i = \prod_{s \in \text{Vertical}(i)} \mathcal{Q}_s, \quad \mathcal{Z}_i = \prod_{p \in \text{Horizontal}(i)} \mathcal{B}_p,$$

The size of the logical operators encodes the property of the code distance, which is seen to be n since there are n independent qubits in each of the definitions of the cyclic operators. We see that the maximum weight of the cyclic operators described above is along the length of the torus, such that we have code distance n , which scales with the square root of the number of qubits encoded.

12.5.2 XZZX Code

While the Toric code provides a scalable, topological approach, it is instructive to also study small-qubit codes that demonstrate different properties. The XZZX code encodes $k = 2 \log$

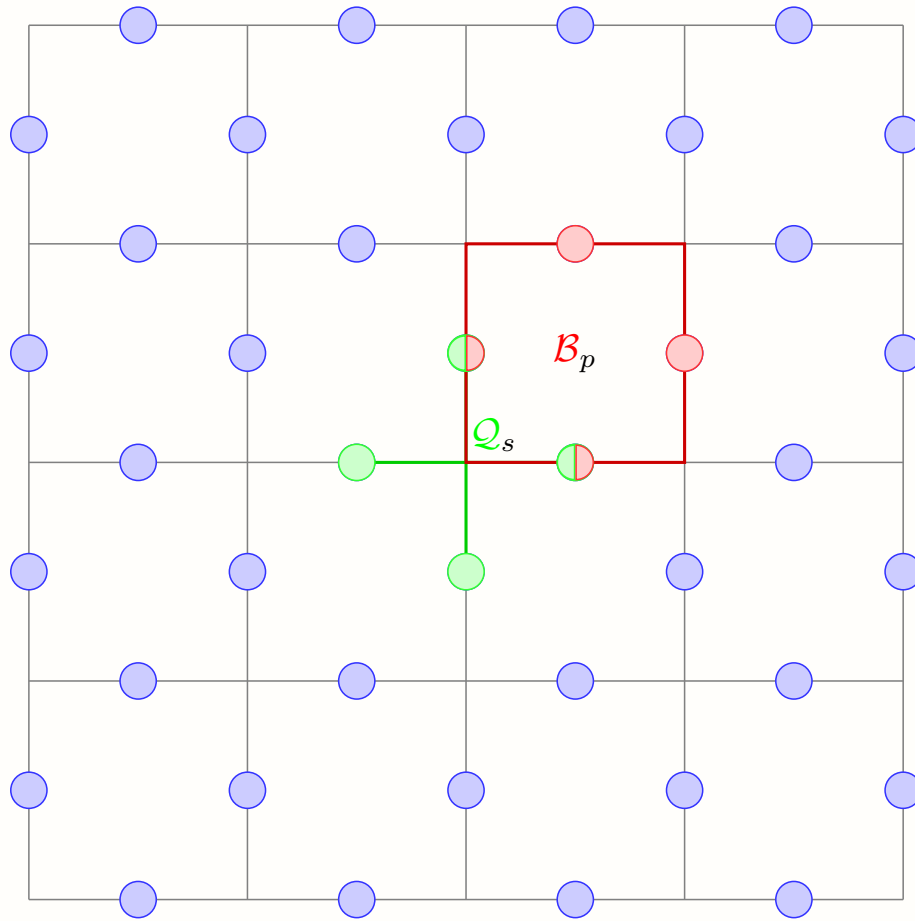


Figure 12.13: Overlapping Toric Code

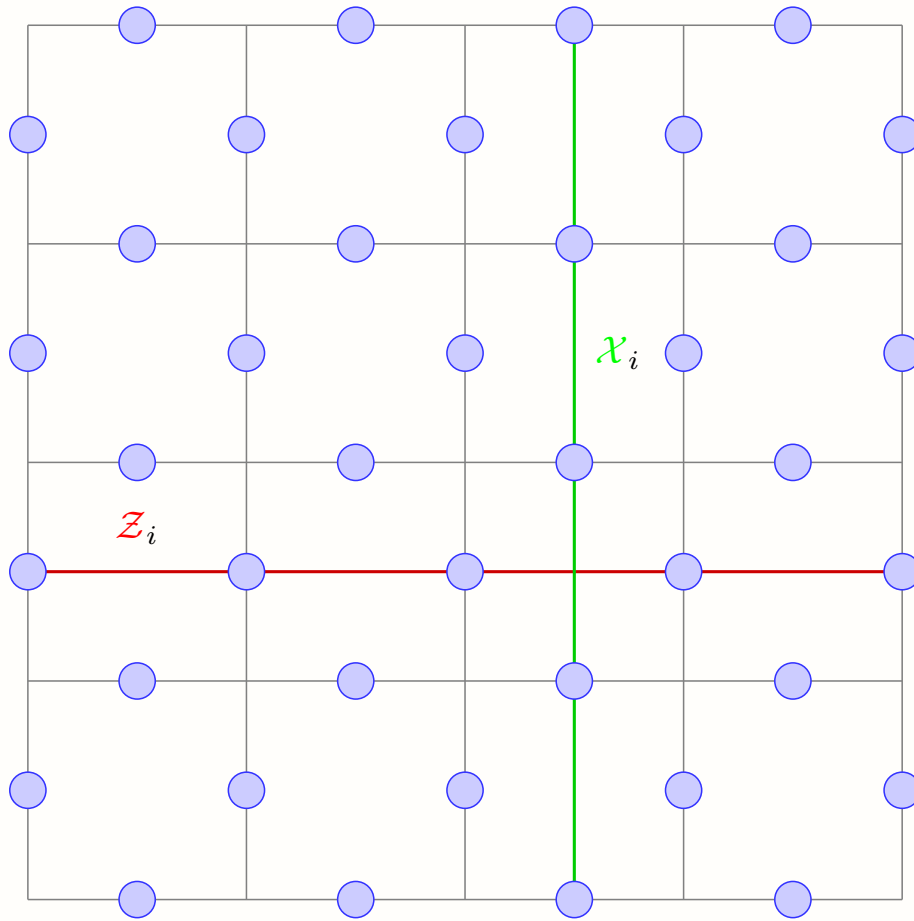


Figure 12.14: Operators on the Torus

ical qubits into $n = 4$ physical qubits, and has a distance $d = 2$.

The code is defined by an Abelian stabilizer group \mathcal{S} generated by $n - k = 4 - 2 = 2$ generators, given by

$$\begin{aligned} M_1 &= X_1 Z_2 Z_3 X_4 \\ M_2 &= Z_1 X_2 X_3 Z_4 \end{aligned}$$

such that $[M_1, M_2] = 0$. The code space C is the 2-qubit ($2^k = 4$) dimensional subspace stabilized by these generators.

The corresponding logical operators (elements of $\mathcal{N}(\mathcal{S}) - \mathcal{S}$) can be defined in pairs. A consistent set is $\bar{X}_1 = X_1 X_2$, $\bar{Z}_1 = Z_1 Z_3$ and $\bar{X}_2 = X_1 X_3$, $\bar{Z}_2 = Z_1 Z_2$.

We must check that these operators behave as logical qubits. First, they must all commute with the stabilizers (which they do, e.g., $[\bar{X}_1, M_1] = [X_1 X_2, X_1 Z_2 Z_3 X_4] = 0$ as X_2 anticommutes with Z_2). Second, they must obey the correct logical commutation relations:

$$[\bar{X}_1, \bar{Z}_1] \neq 0, \quad [\bar{X}_2, \bar{Z}_2] \neq 0 \quad (12.1)$$

$$[\bar{X}_1, \bar{X}_2] = [\bar{Z}_1, \bar{Z}_2] = [\bar{X}_1, \bar{Z}_2] = [\bar{X}_2, \bar{Z}_1] = 0 \quad (12.2)$$

For example, $[\bar{X}_1, \bar{Z}_1] = [X_1 X_2, Z_1 Z_3]$. X_1 anticommutes with Z_1 , while all other pairs commute, resulting in an overall anticommutation, as required. However, $[\bar{X}_1, \bar{Z}_2] = [X_1 X_2, Z_1 Z_2]$. Here, X_1 anticommutes with Z_1 and X_2 anticommutes with Z_2 . With two anticommutations, the operators commute overall.

The minimum weight of a non-trivial logical operator is 2 (e.g., \bar{X}_1), which confirms the code distance $d = 2$.

Further Reading & References

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