

Notes on quantum search algorithm

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Chapter 1

Quantum Computation

This chapter is based on the book[3].

1.1 Quantum Gate

1.1.1 Single-qubit gate

The number of qubits in a quantum state depends on the number of classical bits in its dimension, so we usually call a vector $|\psi\rangle = a|0\rangle + b|1\rangle$ parameterized by two complex numbers a and b satisfying $|a|^2 + |b|^2 = 1$ a qubit. According to the constraints and phase redundancy, the Bloch sphere representation gives us a more intuitive way to observe the qubit, a isomorphism from a qubit to the three-dimensional real sphere:

$$\begin{aligned} \mathbb{CP}^1 &\rightarrow S^2 \\ |\psi\rangle &= \cos \frac{\theta}{2} |0\rangle + \sin \frac{\theta}{2} e^{i\varphi} |1\rangle \mapsto \vec{n} = (\cos \phi \sin \theta, \sin \phi \sin \theta, \cos \theta)^T. \end{aligned} \quad (1.1.1)$$

where $\theta \in [0, \pi]$ and $\psi \in [0, 2\pi]$.

Operations on a qubit must preserve the norm, and thus are described by 2×2 unitary matrices. By $\det |U| = e^{2i\alpha}$ Then

$$e^{-i\alpha}U \in SU(2) \quad (1.1.2)$$

They're only off by one global phase, so we usually ignore the global phase and only consider $SU(2)$. The Lie group $SU(2)$ has three generators $i\sigma_j, \sigma_j$ called Pauli matrices:

$$\sigma_x = \begin{pmatrix} 0 & 1 \\ 1 & 0 \end{pmatrix}, \quad \sigma_y = \begin{pmatrix} 0 & -i \\ i & 0 \end{pmatrix}, \quad \sigma_z = \begin{pmatrix} 1 & 0 \\ 0 & -1 \end{pmatrix}. \quad (1.1.3)$$

From Lie group elements, any single-qubit unitary can be written as a product of exponentials of Pauli matrices by a global phase:

$$U = e^{i\alpha} \exp \left(-i \frac{\omega}{2} \vec{n} \cdot \vec{\sigma} \right), \quad (1.1.4)$$

which \vec{n} is the coordinates of the rotation axis on the Bloch sphere. From the isomorphism 1.1.1, we can also induce a isomorphism of operations on two spaces:

$$\begin{aligned} SU(2)/\{\pm 1\} &\rightarrow SO(3) \\ \exp \left(-i \frac{\omega}{2} \vec{n} \cdot \vec{\sigma} \right) &\mapsto \exp \left(\omega \vec{n} \cdot \vec{J} \right) \end{aligned} \quad (1.1.5)$$

which $\omega \in [0, \pi]$ and J_j is the three generators of Lie group $SO(3)$, the former refers to single-qubit gates and the latter to transformations on the Bloch sphere. For visualization and convenience, we let $R_{\vec{n}}(\omega) := \exp \left(-i \frac{\omega}{2} \vec{n} \cdot \vec{\sigma} \right)$.

In the following, we use XYZ instead of $\sigma_x \sigma_y \sigma_z$, here are some other single-qubit gate which are frequently used, they are called Hadamard gate, phase gate, and $\pi/8$ gate:

$$H = \frac{1}{\sqrt{2}} \begin{pmatrix} 1 & 1 \\ 1 & -1 \end{pmatrix}, \quad S = \begin{pmatrix} 1 & 0 \\ 0 & i \end{pmatrix}, \quad T = \begin{pmatrix} 1 & 0 \\ 0 & e^{i\pi/4} \end{pmatrix}. \quad (1.1.6)$$

1.1.2 Properties of single-qubit gate

Here we list some properties of single-qubit gates:

Theorem 1.1.1. *A single qubit gate U can be decomposed into the following form*

$$U = e^{i\alpha} R_z(\beta) R_y(\gamma) R_z(\delta). \quad (1.1.7)$$

Corollary 1.1.2. *For suitable parameters $\alpha, \beta_k, \gamma_k$ and nonparallel vectors \vec{n} and \vec{m} , there is*

$$U = e^{i\alpha} \prod_{j=1}^k R_{\vec{n}}(\beta_j) R_{\vec{m}}(\gamma_j). \quad (1.1.8)$$

Corollary 1.1.3. *If U is a single qubit gate, then there exists $A, B, C \in SU(2)$, $ABC = I$, and then there is*

$$U = e^{i\alpha} AXBXC \quad (1.1.9)$$

Proof. let

$$A = R_z(\beta) R_y(\gamma/2), \quad B = R_y(-\gamma/2) R_z(-(\gamma + \beta)/2), \quad C = R_z((\gamma - \beta)/2)$$

then we have

$$AXBXC = R_z(\beta) R_y(\gamma) R_z(\delta) \quad (1.1.10)$$

and $ABC = I$. \square

Theorem 1.1.4. *We have the product of pauli matrices,*

$$\sigma_j \sigma_k = i \epsilon_{jk}^l \sigma_l + \delta_{jk} I. \quad (1.1.11)$$

Corollary 1.1.5. *The element of Lie group $SU(2)$ can be changed into the following form,*

$$R_{\vec{n}}(\omega) := \exp\left(-i \frac{\omega}{2} \vec{n} \cdot \vec{\sigma}\right) = \cos\left(\frac{\omega}{2}\right) I - i \sin\left(\frac{\omega}{2}\right) (\vec{n} \cdot \vec{\sigma}). \quad (1.1.12)$$

Proof. Considering the Taylor expansion

$$\begin{aligned} \exp\left(-i \frac{\omega}{2} \vec{n} \cdot \vec{\sigma}\right) &= \sum_{k=0}^{\infty} \frac{1}{k!} \left(-i \frac{\omega}{2} \vec{n} \cdot \vec{\sigma}\right)^k \\ &= \sum_{k=0}^{\infty} \frac{1}{(2k)!} \left(-i \frac{\omega}{2}\right)^{2k} (\vec{n} \cdot \vec{\sigma})^{2k} + \sum_{k=0}^{\infty} \frac{1}{(2k+1)!} \left(-i \frac{\omega}{2}\right)^{2k+1} (\vec{n} \cdot \vec{\sigma})^{2k+1} \\ &= \cos\left(\frac{\omega}{2}\right) I - i \sin\left(\frac{\omega}{2}\right) (\vec{n} \cdot \vec{\sigma}), \end{aligned} \quad (1.1.13)$$

where we have $(\vec{n} \cdot \vec{\sigma})^2 = I$ from the theorem 1.1.4. \square

Corollary 1.1.6. *If a rotation through an angle $2\omega_1$ about an axis \vec{n}_1 is followed by a rotation through an angle $2\omega_2$ about an axis \vec{n}_2 , then the overall rotation is through an angle ω_{12} about an axis \vec{n}_{12} given by*

$$\begin{aligned} \cos \omega_{12} &= \cos \omega_1 \cos \omega_2 - \sin \omega_1 \sin \omega_2 \vec{n}_1 \cdot \vec{n}_2 \\ \sin \omega_{12} \vec{n}_{12} &= \sin \omega_1 \cos \omega_2 \vec{n}_2 + \cos \omega_1 \sin \omega_2 \vec{n}_2 + \sin \omega_1 \sin \omega_2 \vec{n}_1 \times \vec{n}_2. \end{aligned}$$

Proof. We know $R_{\vec{n}_1}(2\omega_1) = \cos \omega I - i \sin \omega (\vec{n}_1 \cdot \vec{\sigma})$ and the same as index 2, so the product of two rotations is

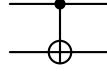
$$\begin{aligned} &R_{\vec{n}_1}(2\omega_1) R_{\vec{n}_2}(2\omega_2) \\ &= \cos \omega_1 \cos \omega_2 I - i \sin \omega_1 \cos \omega_2 (\vec{n}_1 \cdot \vec{\sigma}) - i \cos \omega_1 \sin \omega_2 (\vec{n}_2 \cdot \vec{\sigma}) - \sin \omega_1 \sin \omega_2 (\vec{n}_1 \cdot \vec{\sigma})(\vec{n}_2 \cdot \vec{\sigma}) \\ &= \cos \omega_{12} I - i \sin \omega_{12} (\vec{n}_{12} \cdot \vec{\sigma}), \end{aligned}$$

where we have $(\vec{n}_1 \cdot \vec{\sigma})(\vec{n}_2 \cdot \vec{\sigma}) = i(\vec{n}_1 \times \vec{n}_2) \cdot \vec{\sigma} + (\vec{n}_1 \cdot \vec{n}_2) \cdot \vec{\sigma}$ from Theorem 1.1.4. \square

1.1.3 Controlled operation

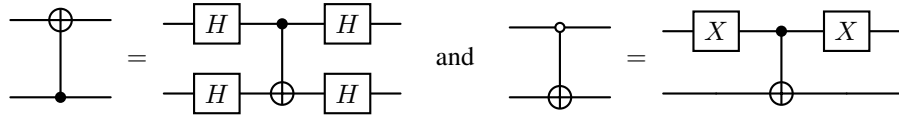
Analogous to the if statement of a classical circuit, we can use controlled-NOT gate to control target qubit with control qubit, its matrix form and circuit are shown as follows:

$$CNOT = \begin{pmatrix} I & 0 \\ 0 & X \end{pmatrix} \quad (1.1.14)$$



$$(1.1.15)$$

CNOT gate makes the state of the target qubit be flipped when the control qubit is 1 and remain unchanged when the control qubit is 0. That is $|c\rangle|t\rangle \rightarrow |c\rangle X^c|t\rangle$. We will also use two deformations of CNOT gates:



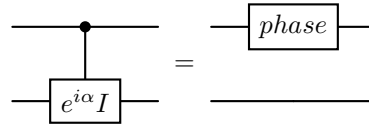
$$(1.1.16)$$

We expect to extend this control from X to an arbitrary single-qubit gate U . That is, $|c\rangle|t\rangle \rightarrow |c\rangle U^c|t\rangle$ which is known as a controlled-U gate. Show that:



$$(1.1.17)$$

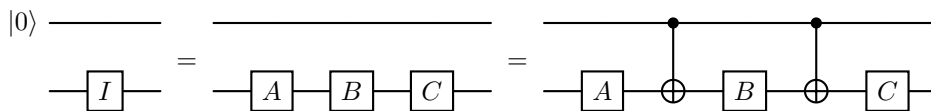
We wish to decompose this into a combination of single quantum bit gates and CNOT gates. With the theorem 1.1.3, we can divide the controlled-U gate into several parts, looking first at the global phase part, which obviously has:



$$(1.1.18)$$

where $phase = \begin{pmatrix} 1 & 0 \\ 0 & e^{i\alpha} \end{pmatrix}$.

Looking at the $AXBXC$ part again, we have the expectation that if the control qubit is $|0\rangle$, then we do nothing, and adding a CNOT gate anywhere doesn't change that, so we have.

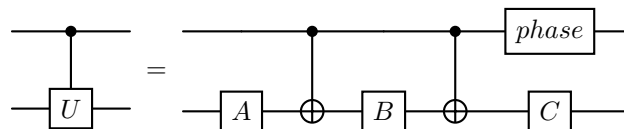


$$(1.1.19)$$

This construction also satisfies: if the control qubit is $|1\rangle$, then we apply $AXBXC$ to the target qubit.

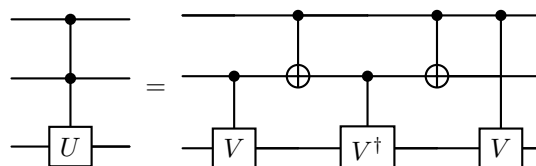
So we have that:

Theorem 1.1.7. *The controlled-U gate can be decomposed with CNOT gates and single qubit gates into this form:*



$$(1.1.20)$$

Considering the case of multiple control Qubits, first look at the case of two control Qubits, you can construct: if V is a single qubit gate satisfying $V^2 = U$, we have:



$$(1.1.21)$$

Then we can extend this to the case of n control qubits, there are many kinds of construction methods, here we use recursive method to construct:

$$(1.1.22)$$

Ultimately we will show that any unitary operation can be composed to an arbitrarily good approximation from just H, S, T, CNOT gates. Here is the construction of the Toffoli gate:

$$(1.1.23)$$

1.2 Universal Gate Set

1.2.1 Single-qubit and CNOT gates are universal

In the section 1.1.3, we showed that $C^n(U)$ gate can be decomposed into a combination of single-qubit gates and CNOT gates. In this section 1.2.1, we will show that any unitary gate can be combined from $C^n(U)$ gate.

First, we introduce the lemma:

Lemma 1.2.1. Any D -dimensional unitary transformation U can always be decomposed into the product of $d(d-1)/2$ two-level unitary transformations under natural basis.

So exists two-level unitary transformations V_j made $U = \prod_{j=1}^{d(d-1)/2} V_j$ ¹. Two-level unitary transformations are unitary

matrices which act non-trivially only on two-or-fewer vector components. The proof idea of this theorem is simple², let's focus on the two-level unitary transformations V .

Consider the binary expansion of the two bases $|s\rangle$ and $|t\rangle$ on which V operates, where $s = s_1 \dots s_n$ and $t = t_1 \dots t_n$. We can use Toffoli gate to turn s_j into t_j , after $n-1$ steps $|s_1 \dots s_{k-1} s_k s_{k+1} \dots s_n\rangle$ into $|t_1 \dots t_{k-1} s_k t_{k+1} \dots t_n\rangle$. It puts $|s\rangle$ and $|t\rangle$ in the same qubit and makes V into a $C^n(\tilde{V})$ gate which \tilde{V} is the non-trivial 2×2 unitary submatrix of V .

Let's take an example to illustrate this lemma, consider the U gate:

$$V = \begin{pmatrix} a & 0 & 0 & c \\ 0 & 1 & 0 & 0 \\ 0 & 0 & 1 & 0 \\ b & 0 & 0 & d \end{pmatrix} \quad \text{and} \quad \tilde{V} = \begin{pmatrix} a & c \\ b & d \end{pmatrix} \quad (1.2.1)$$

Notice that V acts non-trivially only on the states $|00\rangle$ and $|11\rangle$, so we can use the CNOT gate to turn $|00\rangle$ into $|01\rangle$, then we can get the $C(\tilde{V})$ gate. Here is the circuit of V :

$$(1.2.2)$$

So we have:

Theorem 1.2.1. Any unitary gate can be combined from single-qubit gates and CNOT gates.

¹Unless otherwise specified, the conjunction sign in this document is calculated according to the composite order of the map, that is $\prod_{j=1}^n a_j = a_n a_{n-1} \dots a_1$.

²Each single-qubit gates has $d(d-1)/2$ degrees of freedom and each two-level unitary transformations turn one of single-qubit gates elements into zero in order. So exists two-level unitary transformations V_j made $\left(\prod_{j=1}^{d(d-1)/2} V_j \right) U = I$.

1.2.2 Approximate universal operators with a discrete set

It is hard to implement all single-qubit, so we consider to approximate it with a discrete set which we can implement. We introduce the induced norm of operators to measure the degree of approximation:

$$E(U, V) = \|U - V\| := \sup_{\substack{|\psi\rangle \in \mathbb{C}^n \\ \|\psi\|=1}} \|(U - V)|\psi\rangle\| \quad (1.2.3)$$

Where U is the target unitary operator that we wish to implement, V is the unitary operator that is actually implemented in practice, and $E(U, V)$ is the error when V is implemented instead of U . It is natural that:

Theorem 1.2.2. *in the approximation of m gates, the error add at most linearly:*

$$E\left(\prod_{j=1}^m U_j, \prod_{j=1}^m V_j\right) \leq \sum_{j=1}^m E(U_j, V_j) \quad (1.2.4)$$

Proof. Considering $|\psi_0\rangle$ which maximizes $\left\|\left(\prod_{i=1}^m U_i - \prod_{i=1}^m V_i\right)|\psi\rangle\right\|$ and definition $|\psi_i\rangle = V_i|\psi_{i-1}\rangle$, $|\Delta_i\rangle = U_i|\psi_{i-1}\rangle - |\psi_i\rangle$, we can see that

$$E\left(\prod_{j=1}^m U_j, \prod_{j=1}^m V_j\right) = \left\|\Delta_m\right\| + \sum_{j=1}^{m-1} \left\|\left(\prod_{k=j+1}^m U_k\right)|\Delta_j\rangle\right\| \leq \sum_{j=1}^m \|\Delta_j\| \leq \sum_{j=1}^m E(U_j, V_j) \quad \square$$

We are more concerned with the error of the approximation in the measurement. considering M is a POVM element in an arbitrary measurement POVM, and P_U (or P_V) is the probability of obtaining this outcome if U (or V) were performed with the state $|\psi\rangle$, We can prove¹ that:

$$|P_U - P_V| \leq 2E(U, V) \quad (1.2.5)$$

This makes it possible that if we want the probability difference between the approximate line and the ideal line on a certain outcome to be within a tolerance $\Delta > 0$, we only need to ensure that $E(U_j, V_j) \leq \Delta/(2m)$.

From the decomposition given by corollary 1.1.2, we can further care about the degree of approximation of $R_{\vec{n}}(\omega)$. Let's start by introducing a neat lemma :

Lemma 1.2.2. *Considering $\alpha, \omega \in \mathbb{R}/2\pi\mathbb{Z}$, if $\omega/\pi \in \mathbb{R} \setminus \mathbb{Q}$, we can find a subsequences $\{x_n\}$ of sequences $\{n\}$ makes $\lim_{n \rightarrow \infty} x_n \omega = \alpha$.*

Proof. $\forall \epsilon > 0, \exists N = \frac{2\pi}{\epsilon}$, when $n > N$, we can find $|j\omega - k\omega| \leq \frac{2\pi}{n}$ which $j, k \in \{n\}, j > k$, let $x_n = (j-k) \left\lfloor \frac{\alpha}{|j\omega - k\omega|} \right\rfloor$, so we have $|x_n \omega - \alpha| < |j\omega - k\omega| \leq \frac{2\pi}{n} < \epsilon$. \square

Further we have²:

$$\begin{aligned} & E(R_{\vec{n}}(\alpha), R_{\vec{n}}(\omega)^{x_n}) \\ &= E(R_{\vec{n}}(\alpha), R_{\vec{n}}(\alpha + (x_n \omega - \alpha))) \\ &= |1 - \exp((x_n \omega - \alpha)/2)| \leq \epsilon/2 \end{aligned} \quad (1.2.6)$$

According to the proof, we just need to find the gate that has the Angle that the theorem requires. Fortunately, THH gate has the angel we need. It is a rotation of the Bloch sphere about an axis along $\vec{n} = (\cos \frac{\pi}{8}, \sin \frac{\pi}{8}, \cos \frac{\pi}{8})$ and through an angle ω defined by $\cos \frac{\omega}{2} = \cos^2 \frac{\pi}{8}$, so we just let $R_{\vec{n}}(\omega) = THH$.

From corollary 1.1.2, to construct unitary operator U , we still need to approximate the rotation of the other axis, but even more fortunate is that $HR_{\vec{n}}(\omega)H$ is exactly what we need, has axis $\vec{m} = (\cos \frac{\pi}{8}, -\sin \frac{\pi}{8}, \cos \frac{\pi}{8})$ and the angle ω , like $R_{\vec{n}}(\omega)$, is an irrational multiple of π . After layers and layers of preparation, let's list the final approximation with suitable positive integers j_n to U :

$$E(U, \prod_{j=1}^{2k} R_{\vec{n}}(\omega)^{j_n} H) \leq k\epsilon \quad (1.2.7)$$

Theorem 1.2.3. *Given any single qubit unitary operator U and any $\epsilon > 0$, it is possible to approximate U to within ϵ using a circuit composed of H gates and T gates alone.*

¹Let $|\Delta\rangle = (U - V)|\psi\rangle$, notice that $\langle\psi|U^\dagger MU|\psi\rangle - \langle\psi|V^\dagger MV|\psi\rangle = \langle\psi|U^\dagger M|\Delta\rangle + \langle\Delta|MV|\psi\rangle$.

²The construction of x_n in this part is only to prove the existence, without considering the complexity. There are actually far less complex constructs.

³The difference between Equation 1.2.7 and here by one coefficient can be solved by the setting of ϵ in the proof of Theorem 1.2.2, but it does not matter.

1.2.3 Circuit size for approximation

It makes no sense to talk about its existence without giving its size, so let's estimate the number of gates needed for the approximation in the previous section. For convenience of stating the theorem, we say that S is an ϵ -net in W , if every point in W is within a distance ϵ of some point in S , where $S, W \in SU(2)$ and $\epsilon > 0$ and the distance is $D(U, V) := \text{tr}|U - V|$. And we define \mathcal{G}_l to be the set of all words of length at most l .

Theorem 1.2.4 (Solovay-Kitaev theorem). *Let \mathcal{G} be a finite set of elements in $SU(2)$ containing its own inverses, such that $\langle \mathcal{G} \rangle$ is dense in $SU(2)$. Let $\epsilon > 0$ be given. Then \mathcal{G}_l is an ϵ -net in $SU(2)$ for $l = O(\ln^c(1/\epsilon))$, where $c = \ln 5 / \ln(3/2)$.*

According to the theorem, we can know that to approximate a circuit containing m single qubit unitaries to an accuracy ϵ requires $O(m \ln^c(m/\epsilon))$ gates from the discrete set. The proof of this theorem is quite long, so only the main ideas are given here. Let's first introduce the lemma:

Lemma 1.2.3. *Let \mathcal{G} be a finite set of elements in $SU(2)$ containing its own inverses, such that $\langle \mathcal{G} \rangle$ is dense in $SU(2)$. For some $\epsilon > 0$ and any $k \in \mathbb{N}$, if \mathcal{G}_l is an ϵ^2 -net for S_ϵ , then $\mathcal{G}_{5^k l}$ is an $\epsilon(k)^2$ -net for $S_{\epsilon(k)}$, where $\epsilon(k) = (C\epsilon)^{(3/2)^k} / C$, $\epsilon(k)^2 < \epsilon(k+1)$ and $S_\epsilon := \{U \in SU(2) | D(U, I) \leq \epsilon\}$.*

Proof. We know the element of $SU(2)$ can be written as $U = \exp(-i\vec{a} \cdot \vec{\sigma}/2)$, we write $U = u(\vec{a})$. And introduce the symbol $[U, V]_{gp} = UVU^\dagger V^\dagger$. With constant d , we have two conclusions:

$$D([u(\vec{a}), u(\vec{b})]_{gp}, u(\vec{a} \times \vec{b})) \leq d\epsilon^3 \quad \text{and} \quad D(u(\vec{a}), u(\vec{b})) = \|\vec{a} - \vec{b}\| + O(\epsilon^3) \quad (1.2.8)$$

If $U = u(\vec{x}) \in S_{\epsilon^2}$, we can find $\vec{y} \times \vec{z} = \vec{x}$ which $u(\vec{y}), u(\vec{z}) \in S_\epsilon$, so that we can find $u(\vec{y}_0), u(\vec{z}_0) \in \mathcal{G}_l \cap S_\epsilon$ make $D(u(\vec{y}_0), u(\vec{y})), D(u(\vec{z}_0), u(\vec{z})) \leq \epsilon$. Notice that

$$\begin{aligned} D(U, [u(\vec{y}_0), u(\vec{z}_0)]_{gp}) &\leq D(U, u(\vec{y}_0 \times \vec{z}_0)) + D(u(\vec{y}_0 \times \vec{z}_0), [u(\vec{y}_0), u(\vec{z}_0)]_{gp}) \\ &= \|\vec{y} \times \vec{z} - \vec{y}_0 \times \vec{z}_0\| + d\epsilon^3 \\ &\leq (d+2)\epsilon^3 + O(\epsilon^4) \\ &\leq C\epsilon^3 = \epsilon(1)^2 \end{aligned} \quad (1.2.9)$$

Specifically, given $U \in S_{\epsilon(1)}$, we can find $V \in \mathcal{G}_l$ such that $D(U, V) \leq \epsilon(0)^2$, and thus $UV^\dagger \in S_{\epsilon(0)^2}$, so

$$D([u(\vec{y}_0), u(\vec{z}_0)]_{gp}, UV^\dagger) = D([u(\vec{y}_0), u(\vec{z}_0)]_{gp} V, U) \leq \epsilon(1)^2 \quad (1.2.10)$$

that is, \mathcal{G}_{5l} is an $\epsilon(1)^2$ -net for $S_{\epsilon(1)}$. Recursively, $\mathcal{G}_{5^k l}$ is an $\epsilon(k)^2$ -net for $S_{\epsilon(k)}$. \square

For $U \in SU(2)$ we can find U_0 make $D(U_0, U) < \epsilon(0)^2 < \epsilon(1)$, so we have a first order approximation U_0 of U . We can go on to approximate their difference $V = UU_0^\dagger$. With the lemma, we can find U_1 make $D(U_1, V) < \epsilon(1)^2 < \epsilon(2)$, so we have a second order approximation $U_1 U_0$ of U . Continue in this way, we can approximate to the accuracy we want which $\epsilon(k+1) < \epsilon$.

Although the approximation of a set of single-qubit gates is polynomial, approximating arbitrary unitary gates is actually hard. Considering the normalization of n qubit states $\|\psi\| = 1$ gives that the state space is an unit $(2^{n+1} - 1)$ -dimensional unit sphere. Given the error ϵ , a state approximation gives an $(2^{n+1} - 2)$ -dimensional sphere of radius ϵ . So in order to cover the state space, we need about¹

$$\frac{S_{2^{n+1}-1}(1)}{V_{2^{n+1}-2}(\epsilon)} = \frac{\sqrt{\pi}\Gamma(2^n - \frac{1}{2})(2^{n+1} - 1)}{\Gamma(2^n)\epsilon^{2^{n+1}-1}} = \Omega(\epsilon^{-2^{n+1}+1}) \quad (1.2.11)$$

states, where² $S_d(r)$, $V_d(r)$ is the surface area, volume of a d -dimensional sphere of radius r . But for a fixed initial state, m gates can only compute $O(n^{km})$ different states at most, where k is a constant. We must have:

$$O(n^{km}) \geq \Omega(\epsilon^{-2^{n+1}+1}) \quad (1.2.12)$$

which gives us

$$m = \Omega\left(\frac{2^n \ln(1/\epsilon)}{\ln(n)}\right). \quad (1.2.13)$$

This is exponential in n , so it is hard to approximate arbitrary unitary gates.

¹ O is for upper bounds, Ω is for lower bounds and Θ is for tight bounds.

² $S_k(r) = 2\pi^{(k+1)/2} r^k / \Gamma((k+1)/2)$, $V_k(r) = 2\pi^{(k+1)/2} r^{k+1} / (k+1)\Gamma((k+1)/2)$, where $\Gamma(s) = \int_0^\infty t^{s-1} e^{-t} dt$.

1.3 Quantum Circuit Model

Before discussing the algorithm, let's make some basic assumptions about quantum computers (quantum circuit model) clear:

- A quantum computer consists of a classical part and a quantum part: The classical part is unnecessary but can simplify many tasks.
- A quantum circuit operates on n qubits, so the state space is 2^n -dimensional complex Hilbert space.
- It is assumed that any computational basis state $|x_1, \dots, x_n\rangle$ can be prepared in at most n steps.
- Gates can be applied to any subset of qubits as desired, and a universal family of gates can be implemented.
- Measurements may be performed in the computational basis of one or more of the qubits in the computer.

Chapter 2

Quantum Search Algorithm

2.1 Grover's Algorithm

This section is based on the book[3].

2.1.1 Main idea

In the search algorithm, we have the $N = 2^n$ elements in the search space $S = \{0, 1\}^n$ and the target element $T \subseteq S$ which has M elements. The output of the algorithm should be the state containing all the target elements, so we let

$$|\beta\rangle := \frac{1}{\sqrt{M}} \sum_{x \in T} |x\rangle \quad \text{and} \quad |\alpha\rangle := \frac{1}{\sqrt{N-M}} \sum_{x \notin T} |x\rangle. \quad (2.1.1)$$

And the input of the algorithm should be a trivial state, may as well let¹

$$|\psi\rangle := H^{\otimes n} |0\rangle \quad (2.1.2)$$

which use $|0\rangle$ to refer to $|0\rangle^{\otimes n}$ for convenience. There is

$$\langle \alpha | \beta \rangle = 0 \quad \text{and} \quad |\psi\rangle = \sqrt{\frac{N-M}{N}} |\alpha\rangle + \sqrt{\frac{M}{N}} |\beta\rangle. \quad (2.1.3)$$

We need to find some operation makes $|\psi\rangle$ into $|\beta\rangle$. The operation of turning some non-trivial Angle is hard, so we consider the operation reflecting across some axis. Define two operations here, O is a reflection about $|\alpha\rangle$ and W is a reflection about $|\psi\rangle$.

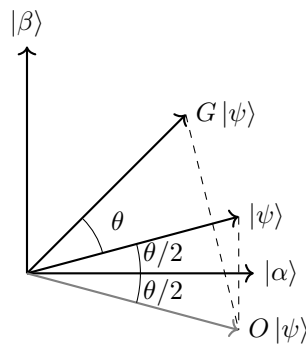


Figure 2.1

Let $\sin(\theta/2) = \sqrt{M/N}$, the angle between $|\psi\rangle$ and $|\alpha\rangle$ is $\theta/2$. As shown in the Figure 2.1, operation $G = WO$ gives us a rotation of angle θ , we call G the Grover iteration. So we just need applying $R := \lfloor \frac{\pi-\theta}{2\theta} \rfloor$ ² times Grover iteration on $|\psi\rangle$ to approximate $|\beta\rangle$, if $M \ll N$ we have

$$G^R |\psi\rangle = \cos\left(\frac{2R+1}{2}\theta\right) |\alpha\rangle + \sin\left(\frac{2R+1}{2}\theta\right) |\beta\rangle \approx |\beta\rangle. \quad (2.1.4)$$

The angular error is at most $\theta/2 \approx \sqrt{M/N}$.

¹ $|\psi\rangle$ is a commonly used state, called the equal superposition state.

²The actual function about $\lfloor x \rfloor$ is to denote the integer closest to the real number x .

2.1.2 Oracle

Let's construct the reflection about $|\alpha\rangle$, we call it oracle. We know that states in the space spanned by $|\alpha\rangle$ and $|\beta\rangle$ can be decomposed into

$$|x\rangle = |\alpha\rangle \langle\alpha|x\rangle + |\beta\rangle \langle\beta|x\rangle \quad (2.1.5)$$

and we wish

$$O|x\rangle = |\alpha\rangle \langle\alpha|x\rangle - |\beta\rangle \langle\beta|x\rangle. \quad (2.1.6)$$

But we don't know the exact form of $|\alpha\rangle$ and $|\beta\rangle$. To construct this, we need a special gate¹

$$\begin{array}{c} |x\rangle \xrightarrow{n} \text{---} x \text{---} \\ |-\rangle \text{---} y \text{---} \end{array} \begin{array}{c} \boxed{U_f} \\ y \oplus f(x) \end{array} \begin{array}{c} x \text{---} \\ |-\rangle \text{---} \end{array} (-1)^{f(x)} |x\rangle \quad (2.1.7)$$

which gives us a operation $|x\rangle \xrightarrow{U_f} (-1)^{f(x)} |x\rangle$. Then we just need to construct the mapping

$$\zeta(x) = \begin{cases} 1 & x \in T \\ 0 & x \notin T \end{cases} \quad (2.1.8)$$

such that U_ζ satisfies the properties we need:

$$\begin{aligned} U_\zeta |\beta\rangle &= -|\beta\rangle \\ U_\zeta |\alpha\rangle &= |\alpha\rangle. \end{aligned} \quad (2.1.9)$$

So the U_ζ is the Oracle.

2.1.3 Grover iteration

Then, let's construct the reflection about $|\psi\rangle$. Same as the oracle, we decompose states into

$$|x\rangle = |\psi\rangle \langle\psi|x\rangle + |\psi_\perp\rangle \langle\psi_\perp|x\rangle \quad (2.1.10)$$

which $|\psi_\perp\rangle = \sqrt{\frac{M}{N}} |\alpha\rangle - \sqrt{\frac{N-M}{N}} |\beta\rangle$, and we wish

$$W|x\rangle = |\psi\rangle \langle\psi|x\rangle - |\psi_\perp\rangle \langle\psi_\perp|x\rangle. \quad (2.1.11)$$

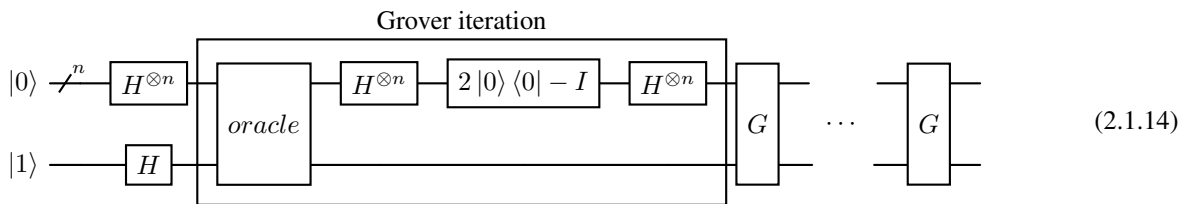
But this time, we know $|\psi\rangle$ specifically. Operation $2|\psi\rangle \langle\psi| - I$ is what we need which satisfies the properties:

$$\begin{aligned} (2|\psi\rangle \langle\psi| - I)|\psi\rangle &= |\psi\rangle \\ (2|\psi\rangle \langle\psi| - I)|\psi_\perp\rangle &= -|\psi_\perp\rangle. \end{aligned} \quad (2.1.12)$$

So we get the construction of the Grover iteration:

$$G = WO = (2|\psi\rangle \langle\psi| - I)U_\zeta. \quad (2.1.13)$$

After constructing each sub-operation, the circuit of the algorithm is given here²:



The ellipsis indicates that it is repeated R times.

¹Note that the mapping $|x\rangle \xrightarrow{U_f} (-1)^{f(x)} |x\rangle$ holds only if the controlled qubit is $|-\rangle$.

²The oracle workspace are omitted here.

2.1.4 Performance

Each of the operations in the Grover iteration may be efficiently implemented on a quantum computer. So we just need to focus on the number of Grover iteration.

We already know, in order to rotate $|\psi\rangle$ near $|\beta\rangle$, we need to repeat the Grover iteration

$$R = \left\lceil \frac{2 \arccos \sqrt{M/N}}{\arcsin \sqrt{M/N}} \right\rceil \quad (2.1.15)$$

times. This is not intuitive enough, so let's sacrifice some precision to find a simpler expression. We know that $R \leq \lceil \pi/2\theta \rceil$, and if $M \leq N/2$ we have

$$\frac{\theta}{2} \geq \sin \frac{\theta}{2} = \sqrt{\frac{N}{M}}, \quad (2.1.16)$$

from which we obtain an elegant upper bound on the number of iterations required,

$$R \leq \left\lceil \frac{\pi}{4} \sqrt{\frac{N}{M}} \right\rceil. \quad (2.1.17)$$

That is, $R = O(\sqrt{N/M})$ Grover iterations must be performed in order to obtain a solution to the search problem with high probability, a quadratic improvement over the $O(N/M)$ oracle calls required classically.

If $M > N/2$, the quantum search algorithm is not necessary. We can just randomly pick an item from the search space, and then check that it is a solution using the oracle. This approach has a success probability at least one-half, and only requires one consultation with the oracle.

2.2 Bounds for Grover's algorithm

This section is based on the paper[1].

2.2.1 Main idea

Oracle is at the heart of grover's algorithm, so let's focus on the oracle and ignore the other operations, we write the state which uses R queries to the oracle as

$$|\Psi_x^R\rangle = \prod_{j=1}^R U_j O_x |\psi\rangle, \quad (2.2.1)$$

where U_i is some unitary operation and O_x is the oracle which marks the target element x . To illustrate the effect of oracles, let's replace some of them with unit operations, we write

$$|\Psi_x^{i,R}\rangle = \prod_{j=i+1}^R U_j O_x \prod_{j=1}^i U_j I |\psi\rangle, \quad (2.2.2)$$

which use the identity for the first i oracle queries and the oracle for the latter $R-i$ oracle queries. If there is no application oracle, then the state is independent of the target element, may be denoted as

$$|\Psi^R\rangle = \prod_{j=1}^R U_j I |\psi\rangle. \quad (2.2.3)$$

So that we can find the distance changed by oracle, here we use two kinds of metric, euclidean distance and angular distance:

$$E(|\alpha\rangle, |\beta\rangle) = \|\alpha - \beta\| \quad \text{and} \quad A(|\alpha\rangle, |\beta\rangle) = \arccos \frac{|\langle\alpha|\beta\rangle|}{\|\alpha\| \|\beta\|}. \quad (2.2.4)$$

Taking the distance E as an example, if Ψ_x^R satisfies the properties the output should have

$$\|\Pi_x |\Psi_x^R\rangle\|^2 \geq p \quad (2.2.5)$$

which $\{\Pi_j\}$ is a finite set of orthogonal projectors that sum to the identity and p is the probability we want of outputting the correct element.

We can calculate the number of queries R by dividing $E(|\Psi_x^R\rangle, |\Psi^R\rangle)$ with the distance changed by oracle¹, that is

$$R = \frac{E(|\Psi_x^R\rangle, |\Psi^R\rangle)}{E(O_x |\gamma\rangle, |\gamma\rangle)}. \quad (2.2.6)$$

It is important to note that $|\gamma\rangle$ here is not some particular state, we will calculate the mean of the distance changed by oracle on each state.

¹Note that this is just a loose intuition, which we'll prove in the next subsection.

2.2.2 Distance the oracle changes

First, let's calculate the distance changed by oracle. Considering the distance of the state before and after R oracle queries, we have

Lemma 2.2.1. *The average **euclidean distance** after R oracle queries is at most*

$$\frac{1}{N} \sum_{x=1}^N E(|\Psi_x^R\rangle, |\Psi^R\rangle) \leq 2R \frac{1}{\sqrt{N}}. \quad (2.2.7)$$

Proof. With triangle inequality,

$$\begin{aligned} \frac{1}{N} \sum_{x=1}^N E(|\Psi_x^R\rangle, |\Psi^R\rangle) &= \frac{1}{N} \sum_{x=1}^N E(|\Psi_x^{0,R}\rangle, |\Psi_x^{R,R}\rangle) \leq \frac{1}{N} \sum_{x=1}^N \sum_{i=1}^R E(|\Psi_x^{i-1,R}\rangle, |\Psi_x^{i,R}\rangle) \\ &= \frac{1}{N} \sum_{i=1}^R \sum_{x=1}^N E(O_x |\Psi^i\rangle, |\Psi^i\rangle) = \frac{1}{N} \sum_{i=1}^R \sum_{x=1}^N 2 \|\Pi_x |\Psi^i\rangle\| \\ &\leq 2 \sum_{i=1}^R \frac{1}{\sqrt{N}} = 2R \frac{1}{\sqrt{N}}, \end{aligned}$$

where the last inequality needs to notice that

$$\left(\sum_{i=1}^N a_i \right)^2 = \sum_{i=1}^N a_i^2 + 2 \sum_{i=1}^N \sum_{j=1}^{i-1} a_i a_j \leq N \sum_{i=1}^N a_i^2, \quad (2.2.8)$$

that is

$$\sum_{x=1}^N \frac{1}{N} \|\Pi_x |\Psi^i\rangle\| \leq \sqrt{\frac{1}{N} \sum_{x=1}^N \|\Pi_x |\Psi^i\rangle\|^2} = \frac{1}{\sqrt{N}}. \quad \square$$

Lemma 2.2.2. *The average **angular distance** after R oracle queries is at most*

$$\frac{1}{N} \sum_{x=1}^N A(|\Psi_x^R\rangle, |\Psi^R\rangle) \leq 2R \arcsin\left(\frac{1}{\sqrt{N}}\right). \quad (2.2.9)$$

Proof. With triangle inequality,

$$\begin{aligned} \frac{1}{N} \sum_{x=1}^N A(|\Psi_x^R\rangle, |\Psi^R\rangle) &= \frac{1}{N} \sum_{x=1}^N A(|\Psi_x^{0,R}\rangle, |\Psi_x^{R,R}\rangle) \leq \frac{1}{N} \sum_{x=1}^N \sum_{i=1}^R A(|\Psi_x^{i-1,R}\rangle, |\Psi_x^{i,R}\rangle) \\ &= \frac{1}{N} \sum_{i=1}^R \sum_{x=1}^N A(O_x |\Psi^i\rangle, |\Psi^i\rangle) = \frac{1}{N} \sum_{i=1}^R \sum_{x=1}^N \arccos(|\cos(2\theta_x^i)|) \\ &\leq 2 \sum_{i=1}^R \arcsin\left(\frac{1}{\sqrt{N}}\right) = 2R \arcsin\left(\frac{1}{\sqrt{N}}\right), \end{aligned}$$

where has $\theta_x^i = A(|\Psi^i\rangle, |x_\perp\rangle) = \arcsin \|\Pi_x |\Psi^i\rangle\|$. And where the last inequality still needs to notice the rescaling provided by equation 2.2.8, that is

$$\sum_{x=1}^N \frac{1}{N} \arcsin \|\Pi_x |\Psi^i\rangle\| \leq \arcsin \sum_{x=1}^N \frac{1}{N} \|\Pi_x |\Psi^i\rangle\| \leq \arcsin \frac{1}{\sqrt{N}},$$

where also need notice that $\sum_{x=1}^N \frac{1}{N} \arcsin a_i \leq \arcsin \sum_{x=1}^N \frac{1}{N} a_i$ which $a_i \in [0, \frac{\pi}{2}]$. \square

According to the theorem, the distance changed by each oracle query can only add up linearly.

2.2.3 Distance when the output is we need

Then, let's calculate the distance when equation 2.2.5 is satisfied.

Lemma 2.2.3. *Suppose that the algorithm correctly outputs y with probability at least p after R queries, given oracle O_y . Then the average euclidean distance is at least*

$$\frac{1}{N} \sum_{x=1}^N E(|\Psi_x^R\rangle, |\Psi^R\rangle) \geq \frac{1}{\sqrt{2}} \left(\sqrt{p} - \sqrt{1-p} + 1 - \frac{2}{\sqrt{N}} \right). \quad (2.2.10)$$

Proof. For a certain x , we have

$$\begin{aligned} E(|\Psi_x^R\rangle, |\Psi^R\rangle) &\geq \frac{1}{\sqrt{2}} (\|\Pi_x (|\Psi_x^R\rangle - |\Psi^R\rangle)\| + \|\Pi_x^\perp (|\Psi_x^R\rangle - |\Psi^R\rangle)\|) \\ &\geq \frac{1}{\sqrt{2}} (\|\Pi_x |\Psi_x^R\rangle\| - \|\Pi_x |\Psi^R\rangle\| + \|\Pi_x^\perp |\Psi_x^R\rangle\| - \|\Pi_x^\perp |\Psi^R\rangle\|) \\ &\geq \frac{1}{\sqrt{2}} (\|\Pi_x |\Psi_x^R\rangle\| - \|\Pi_x^\perp |\Psi_x^R\rangle\| + 1 - 2\|\Pi_x |\Psi^R\rangle\|) \\ &\geq \frac{1}{\sqrt{2}} \left(\sqrt{p} - \sqrt{1-p} + 1 - 2\|\Pi_x |\Psi^R\rangle\| \right), \end{aligned}$$

where the last inequality from the success probability being at least p . To eliminate the last term, consider a traversal of x

$$\frac{1}{N} \sum_{x=1}^N E(|\Psi_x^R\rangle, |\Psi^R\rangle) \geq \frac{1}{\sqrt{2}} \left(\sqrt{p} - \sqrt{1-p} + 1 - \frac{2}{N} \sum_{x=1}^N \|\Pi_x |\Psi^R\rangle\| \right) \geq \frac{1}{\sqrt{2}} \left(\sqrt{p} - \sqrt{1-p} + 1 - \frac{2}{\sqrt{N}} \right),$$

where the last inequality uses the equation 2.2.8 again. \square

Lemma 2.2.4. *Suppose that the algorithm correctly outputs y with probability at least p after R queries, given oracle O_y . Then the average angular distance is at least*

$$\frac{1}{N} \sum_{x=1}^N A(|\Psi_x^R\rangle, |\Psi^R\rangle) \geq \arcsin \sqrt{p} - \arcsin \frac{1}{\sqrt{N}}. \quad (2.2.11)$$

Proof. For a certain x , we have

$$\begin{aligned} A(|\Psi_x^R\rangle, |\Psi^R\rangle) &= \arccos (|\langle \Psi_x^R | \Psi^R \rangle|) \\ &= \arccos (|\langle \Psi_x^R | (\Pi_x + \Pi_x^\perp)^\dagger (\Pi_x + \Pi_x^\perp) | \Psi^R \rangle|) \\ &= \arccos (\|\Pi_x |\Psi_x^R\rangle\| \cdot \|\Pi_x |\Psi^R\rangle\| + \|\Pi_x^\perp |\Psi_x^R\rangle\| \cdot \|\Pi_x^\perp |\Psi^R\rangle\|) \\ &= \arccos (\sin \phi_x^R \sin \theta_x^R + \cos \phi_x^R \cos \theta_x^R) \\ &= \phi_x^R - \theta_x^R \geq \arcsin \sqrt{p} - \theta_x^R, \end{aligned}$$

where has $\phi_x^i = \arcsin \|\Pi_x |\Psi_x^i\rangle\|$ and still has $\theta_x^i = \arcsin \|\Pi_x |\Psi^i\rangle\|$, and where the last inequality from the success probability being at least p . Just like the last proof, consider a traversal of x

$$\frac{1}{N} \sum_{x=1}^N A(|\Psi_x^R\rangle, |\Psi^R\rangle) \geq \arcsin \sqrt{p} - \frac{1}{N} \sum_{x=1}^N \theta_x^R \geq \arcsin \sqrt{p} - \arcsin \frac{1}{\sqrt{N}},$$

where the last inequality uses the equation 2.2.8 again. \square

2.2.4 Lower bound of grover iterations

From section 2.2.2 and section 2.2.3, we obtain the two parts required for equation 2.2.6 to compute the number of iterations: the upper bound of the distance changed by oracle and the lower bound of the distance when the output is we need. So we have the lower bound on the number of iterations for both versions

Theorem 2.2.1. *The unordered search problem with success probability $p > 0$ requires at least R queries, where*

$$R = \frac{\sqrt{N}}{2\sqrt{2}} \left(\sqrt{p} - \sqrt{1-p} + 1 - \frac{2}{\sqrt{N}} \right). \quad (2.2.12)$$

Theorem 2.2.2. *The unordered search problem with success probability $p > 0$ requires at least R queries, where*

$$R = \frac{\arcsin \sqrt{p} - \arcsin \frac{1}{\sqrt{N}}}{2 \arcsin \frac{1}{\sqrt{N}}}. \quad (2.2.13)$$

In the case of euclidean distances, we conclude that Grover's algorithm is asymptotically optimal, and in the case of angular distances, that Grover's algorithm is exactly optimal.

2.3 Long algorithm

This section is based on the paper[2].

2.3.1 Main idea

Although the grover's algorithm is already accurate enough when the number of elements is large, for instance in deciphering the DES code where $N = 256$ the deviation is only 3×10^{-9} . But in problems where certainty is vital, especially when the dimension is not so big, using a searching algorithm with certainty becomes important. Let's make some improvements to the grover's algorithm.

In the grover's algorithm¹ we construct rotations in the space spanned by $|\alpha\rangle$ and $|\beta\rangle$, but $SU(2)$ has three degrees of freedom, so we may as well extend $G = WO = (I - 2|\psi_\perp\rangle\langle\psi_\perp|)(I - 2|\beta\rangle\langle\beta|)$ to

$$\begin{aligned} I_W &= I + (e^{-i\phi} - 1)|\psi_\perp\rangle\langle\psi_\perp| = e^{-i\phi}(I + (e^{i\phi} - 1)|\psi\rangle\langle\psi|) \\ I_O &= I + (e^{i\phi} - 1)|\beta\rangle\langle\beta| \\ Q &= I_W I_O = e^{-i\phi} H^{\otimes n} (I + (e^{i\phi} - 1)|0\rangle\langle 0|) H^{\otimes n} (I + (e^{i\phi} - 1)|\beta\rangle\langle\beta|). \end{aligned} \quad (2.3.1)$$

Intuitively, if we want Q to make $|\psi\rangle$ go to $|\beta\rangle$, we should have the axis of rotation at the same angle as $|\psi\rangle$ and $|\beta\rangle$. From corollary 1.1.5, if the angle of I_O is $-\phi$, the angle of I_W has to be ϕ . As a generalization of the grover's algorithm, it can be seen that if we set $\phi = \pi$, then Q degenerates to G .

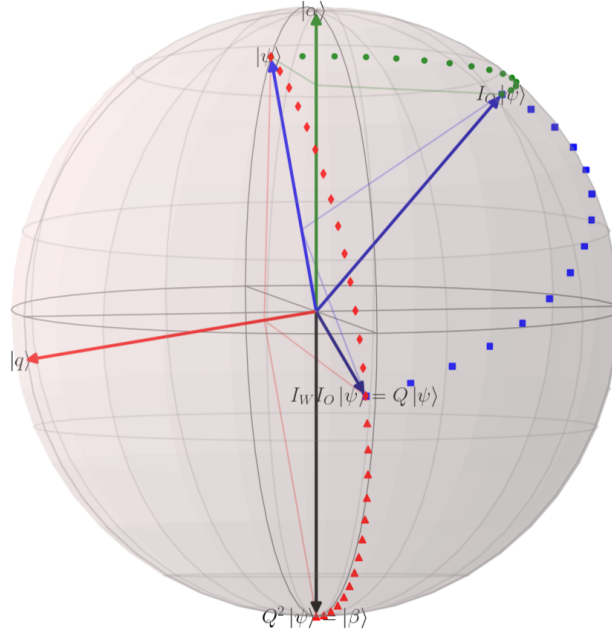


Figure 2.2

As shown in the Figure 2.2, I_O gives us a rotation of angle $-\phi$ about axis $|\alpha\rangle \rightarrow (0, 0, 1)^T$ and I_W gives us a rotation of angle ϕ about axis $|\psi\rangle \rightarrow (\sin \theta, 0, \cos \theta)^T$, their product $Q = I_W I_O$ gives us a rotation of angle $\omega = 4 \arcsin(\sin \frac{\phi}{2} \sin \frac{\theta}{2})$ about axis $|q\rangle \rightarrow k(\cos \frac{\phi}{2}, \sin \frac{\phi}{2}, \cos \frac{\phi}{2} \tan \frac{\theta}{2})^T$, where k is the normalization factor.

¹The notation of section 2.1.1 is followed here.

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