



UNIVERSITÀ DEGLI STUDI DI MILANO
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Master degree in Physics

**Development of an open-source calibration framework for
superconducting qubits**

Supervisor:

Prof. Dr. Stefano Carrazza

Co-supervisor:

Dr. Alessandro Candido

Co-supervisor:

Dr. Andrea Pasquale

Co-supervisor:

Dott. Edoardo Pedicillo

Elisa Stabilini
Matricola n° 28326A
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Summary

Chapter 1

Superconducting qubits

The electronics that modern computers rely on contain components that operate based on quantum mechanics; however, their computational processes are still governed by classical laws. For this reason, they are referred to as "classical computers."

Quantum computing emerged from Richard Feynman's idea that simulating quantum systems efficiently requires quantum mechanical resources [1]. Classical computers struggle to model complex quantum interactions due to the exponential growth of computational requirements with system size, making exact simulations infeasible beyond small systems [2]. Quantum computers, taking advantage of quantum mechanics phenomena like superposition and entanglement, offer a natural framework for such simulations and have been demonstrated to provide exponential speedups for certain quantum systems [3].

Beyond quantum simulation, current theoretical advancements suggest that quantum algorithms can outperform classical counterparts in solving specific problems [4].

1.1 Introduction

The physical realization of quantum computing necessitates the development of a system capable of functioning as quantum bits (qubits).

Similarly to classical logic, where the bits 0 and 1 are associated with two physical levels, typically represented by high and low voltage states, a qubit can, to a first approximation, be considered as a two-level physical system.

Mathematically, this system is described with a two-dimensional complex Hilbert space, where the basis states $|0\rangle$ and $|1\rangle$ correspond to two orthonormal vectors. Any general state of the qubit can be expressed as a superposition of these basis states:

$$|\psi\rangle = \alpha|0\rangle + \beta|1\rangle \rightarrow \begin{pmatrix} \alpha \\ \beta \end{pmatrix}, \quad (1.1)$$

where $\alpha, \beta \in \mathbb{C}$. If the normalization condition $|\alpha|^2 + |\beta|^2 = 1$ holds, the state $|\psi\rangle$ represents a qubit. The basis $\{|0\rangle, |1\rangle\}$ is called computational basis and the information is stored in the complex numbers α and β .

A possible geometric representation of qubit states is given by the Bloch sphere, which offers a visualization of two level quantum systems as vectors on a unit sphere. A qubit state is depicted as a vector originating from the center of the sphere, with the computational basis states $|0\rangle$ and $|1\rangle$ positioned at the north and south poles, respectively. The axis connecting these states defines the z -axis. The transverse x - and y - axes correspond to the equal superposition states $|\pm\rangle = \frac{1}{\sqrt{2}}(|0\rangle \pm |1\rangle)$ and $|\pm i\rangle = \frac{1}{\sqrt{2}}(|0\rangle \pm i|1\rangle)$, respectively.

A vector of unit length on the Bloch sphere is characterized by the polar angle θ , with $0 \leq \theta \leq \pi$ and the azimuthal angle φ , with $0 \leq \varphi \leq 2\pi$, each unit vector represent a possible pure state of the qubit.

The qubit states $|0\rangle$ and $|1\rangle$ can also be associated with energy eigenstates of a physical system, where $|0\rangle$ represents the ground state with energy E_0 and $|1\rangle$ represents the excited state with energy

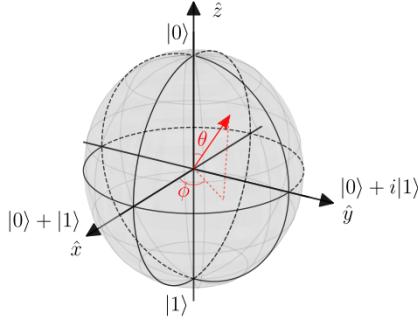


Figure 1.1: Example of qubit state representation on the Bloch sphere

Source: Metrology of Quantum Control and Measurement in Superconducting Qubits [5]

E_1 , assuming $E_0 < E_1$. In this energy eigenbasis, the Hamiltonian of the qubit is given by

$$\hat{H}_q = E_0 |0\rangle\langle 0| + E_1 |1\rangle\langle 1| = \begin{pmatrix} E_0 & 0 \\ 0 & E_1 \end{pmatrix}. \quad (1.2)$$

Since only energy differences are physically relevant, it is possible to redefine the zero-point energy by subtracting the constant term $E_0(|0\rangle\langle 0| + |1\rangle\langle 1|)$, leading to the simplified Hamiltonian

$$\hat{H}_q = (E_1 - E_0) |1\rangle\langle 1| = \hbar\omega_q |1\rangle\langle 1| = \hbar\omega_q \hat{\sigma}^+ \hat{\sigma}^- = \begin{pmatrix} 0 & 0 \\ 0 & \hbar\omega_q \end{pmatrix}, \quad (1.3)$$

where $\omega_q = (E_1 - E_0)/\hbar$ is the qubit transition frequency, and was used the relation $\hat{\sigma}^+ \hat{\sigma}^- = |1\rangle\langle 1|$. For convenience, the Hamiltonian can also be rewritten in terms of the Pauli z -matrix, $\hat{\sigma}_z$, by adding a term proportional to the identity:

$$\hat{H}_q = \hbar\omega_q |1\rangle\langle 1| - \frac{\hbar\omega_q}{2} \mathbb{I} = \begin{pmatrix} -\frac{\hbar\omega_q}{2} & 0 \\ 0 & \frac{\hbar\omega_q}{2} \end{pmatrix} = -\frac{\hbar\omega_q}{2} \hat{\sigma}_z. \quad (1.4)$$

Qubits can be implemented through various physical mechanisms; however, their practical realization remains a significant challenge due to their susceptibility to environmental interactions, which lead to decoherence and reduce their coherence time. Despite the diversity of possible physical implementations, any functional quantum computing system must satisfy a set of fundamental criteria. These requirements, known as the DiVincenzo criteria, establish the essential conditions for the construction and operation of a viable quantum computer [6], [7]:

1. The physical system used as quantum computer must comprise a set of qubits, meaning that the quantum system must be well-characterized, and scalable such that quantum computing can be realized.
2. It must be possible to initialize the qubits in a reliable state, such as the ground state.
3. The coherence time of the qubits must be longer than the typical gate time, ideally should be possible to perform $> 10^4$ operations, that is the number which allows for realizing effective error corrections.
4. It must be possible to implement a universal set of quantum gates.
5. It must be possible to measure the qubits in the computational basis.

In the present work, I will focus on superconducting qubits, which constitute the hardware I have worked on and where the experiments were conducted. However, several of the experiments described later can also be implemented using different physical systems.

1.2 Transmon qubits

In this section, I provide a review of the structure and operation of superconducting transmon qubits. The content of this section is based on the *Quantum Information Science* manual [7], *The Metrology of Quantum Control and Measurement in Superconducting Qubits* [5], the notes from quantum computing lectures held by Professor Olivares [8], and the original transmon paper [9].

1.2.1 Josephson Junctions

The Josephson junction (JJ) is formed by a thin oxide layer positioned between the two superconductors which acts as an insulating barriers. An example of Josephson junction is show in figure (1.2), a side view in image (1.2a) and a top view in image (1.2b).



Figure 1.2: Figure (1.2a): Side viwe of a Josephson junction, the two superconducting pads are coloured in red and blue and indicating by the letter S. In grey, indicated by letter I is represented the insulating barrier of oxide. The superconductors and the oxide are layered over a substrate.

Figure (1.2b): Electron microscope image of a $2\mu\text{m} \times 2\mu\text{m}$ cross-type junction: I. Josephson junction. II. Base electrode. III. Contact to the top electrode.

Source:<https://www.ims.kit.edu/english/2551.php>

Superconductivity is a phenomenon observed in certain materials where, when cooled well below a critical temperature T_c , which depends on the material, their electrical resistance drops to zero, allowing them to behave as perfect conductors. According to the BCS (Bardeen-Cooper-Schrieffer) theory, superconductivity arises, from the formation of Cooper pairs, which are bound states of electrons with opposite momenta and spins. These pairs collectively forms a macroscopic quantum states described by a single waveform $\psi(\mathbf{r})$ which can be expressed as

$$\psi(\mathbf{r}) = \sqrt{\rho(\mathbf{r})} e^{i\theta(\mathbf{r})}, \quad (1.5)$$

where $\rho(\mathbf{r})$ is the density of Cooper pairs in the metal, which is typically uniform in the bulk of the superconductor, and $\theta(\mathbf{r})$ is the macroscopic phase of the superconducting wavfunction.

For this reason the wavefunctions on the two sides of the JJ can be denoted as

$$\psi_1(\mathbf{r}, t) = \sqrt{\rho_1(\mathbf{r}, t)} e^{i\theta_1(\mathbf{r}, t)}, \quad \psi_2(\mathbf{r}, t) = \sqrt{\rho_2(\mathbf{r}, t)} e^{i\theta_2(\mathbf{r}, t)}. \quad (1.6)$$

The dynamics of the system can be described by the two equations

$$i\hbar \frac{d\psi_1}{dt} = E_1 \psi_1 + K \psi_2, \quad (1.7)$$

$$i\hbar \frac{d\psi_2}{dt} = E_2 \psi_2 + K \psi_1. \quad (1.8)$$

By substituting the expression of ψ_i into the Schrödinger equation (1.7), (1.8) we obtain

$$\frac{d\rho_1}{dt} = \frac{2K}{\hbar} \sqrt{\rho_1 \rho_2} \sin(\theta_2 - \theta_1), \quad (1.9)$$

$$\frac{d\rho_2}{dt} = -\frac{2K}{\hbar} \sqrt{\rho_1 \rho_2} \sin(\theta_2 - \theta_1). \quad (1.10)$$

Since the derivative of the charge density is the current, from equations (1.9) and (1.10) we obtain the first Josephson equation

$$I = I_c \sin \phi, \quad (1.11)$$

where $I_c = \frac{2K}{\hbar} \sqrt{\rho_1 \rho_2}$ is the critical current and ϕ is the superconducting phase difference $\theta_2 - \theta_1$.

Instead, from the real part of the Schrödinger equation (1.7), (1.8) and a few calculations, we obtain the second Josephson equation

$$\frac{d\phi}{dt} = \frac{2e}{\hbar} V(t). \quad (1.12)$$

which can be rewritten as

$$\frac{d\phi}{dt} = \frac{2\pi}{\Phi_0} V(t). \quad (1.13)$$

where $\Phi_0 = \frac{h}{2e}$ is the superconducting flux quantum, with h is the Planck's constant and $2e$ is the charge of a Cooper pair.

The time derivative of the first Josephson equation (1.11) yields:

$$\dot{I}_J = I_C \cos \phi \frac{\partial \phi}{\partial t}, \quad (1.14)$$

equation (1.14) suggests a nonlinear relation between the current the voltage. Using the Josephson voltage-phase relation and the fact that $\dot{I} = \frac{V}{L}$ it is possible to define an effective nonlinear inductance for the Josephson junction:

$$L_J = \frac{1}{\cos \phi} \frac{\Phi_0}{2\pi I_c}. \quad (1.15)$$

In addition to the inductive behaviour the Josephson junction also exhibits capacitive properties due to its inherent capacitance C_J with a corresponding energy of

$$E_{C_J} = \frac{Q^2}{2C_J}, \quad (1.16)$$

From equation (1.15) it is possible to compute the energy stored in the nonlinear inductance as

$$E_{L_J} = \int_0^t d\tau I_J(\tau) V(\tau) = \int_0^t d\tau I_c \sin \phi(\tau) \frac{\partial \phi(\tau)}{\partial \tau} \frac{\Phi_0}{2\pi} \quad (1.17)$$

$$= \frac{\Phi_0 I_c}{2\pi} (1 - \cos \phi) = E_J (1 - \cos \phi), \quad (1.18)$$

where E_J represents the energy due to the behaviour of the junction as nonlinear inductor .

1.2.2 CPB qubit

A first example of superconducting qubit is the Cooper Pair Box (CPB), which consists of a small superconducting island connected to a reservoir of superconducting electrons through a Josephson junction [10], with an external gate voltage controlling the charge state. The circuit corresponding to CPB is similar to the circuit of a parallel resonator where the linear inductance is substituted by a Josephson junction which simply acts as a nonlinear inductance.

Combining the energy associated to the capacitance C and the energy of the Josephson junction (1.17) it is possible to write the classical Hamiltonian of the circuit

$$H_J = 4E_C n^2 - E_J \cos \phi, \quad (1.19)$$

where the constant term was ignored as it acts simply as a constant offset without influencing the dynamics of the system and where E_C is the charging energy defined as

$$E_C = \frac{e^2}{2C}. \quad (1.20)$$

To control the number of Cooper pairs on the island, it is possible to connect a DC voltage source V_g to the system through a gate capacitor C_g , as shown in Figure (1.3b). When $V_g = 0$, both the gate and qubit capacitors remain uncharged. As V_g increases, a charge $Q_g = C_g V_g$ accumulates on the gate capacitor, inducing an equal and opposite charge on the island to maintain charge neutrality. When $Q_g \approx 2e$, a Cooper pair tunnels from the reservoir to the island, discharging the qubit capacitor.

The presence of the external voltage source introduces an additional control parameter for the number of Cooper pairs on the island, modifying the system's Hamiltonian. The resulting Hamiltonian of the Cooper Pair Box (CPB) takes the form

$$\hat{H} = 4E_C (\hat{n} - n_g)^2 - E_J \cos \hat{\phi}, \quad (1.21)$$

where $n_g = \frac{C_g V_g}{2e}$ represents the normalized gate charge.

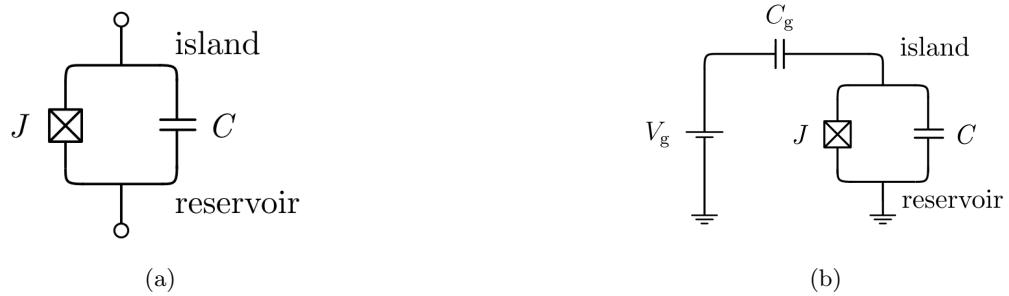


Figure 1.3: Figure (1.3a): corresponding circuit of a CPB which consists of Josephson junction shunted by a capacitor. Source: [7]. Figure (1.3b): electrical circuit of a CPB capacitively coupled to a DC voltage source through a capacitor C_g . Source: [7].

A key feature of the system is the presence of a Josephson junction in place of a linear inductance. Unlike a standard LC circuit—corresponding to a quantum harmonic oscillator—this results in a non-equidistant energy level spectrum. In particular, the energy levels are anharmonic, allowing the first two levels to be spectrally isolated from the higher ones. This anharmonicity enables the use of the subspace spanned by the ground state $|0\rangle$ and the first excited state $|1\rangle$ as a qubit.

The qubit frequency is defined as the frequency associated with the energy difference between these two states: $f_{01}(n_g) = f_q = \frac{(E_1 - E_0)}{\hbar}$. This frequency can be tuned by varying the externally applied DC voltage, which modifies the system's parameters and, consequently, its energy level spacing.

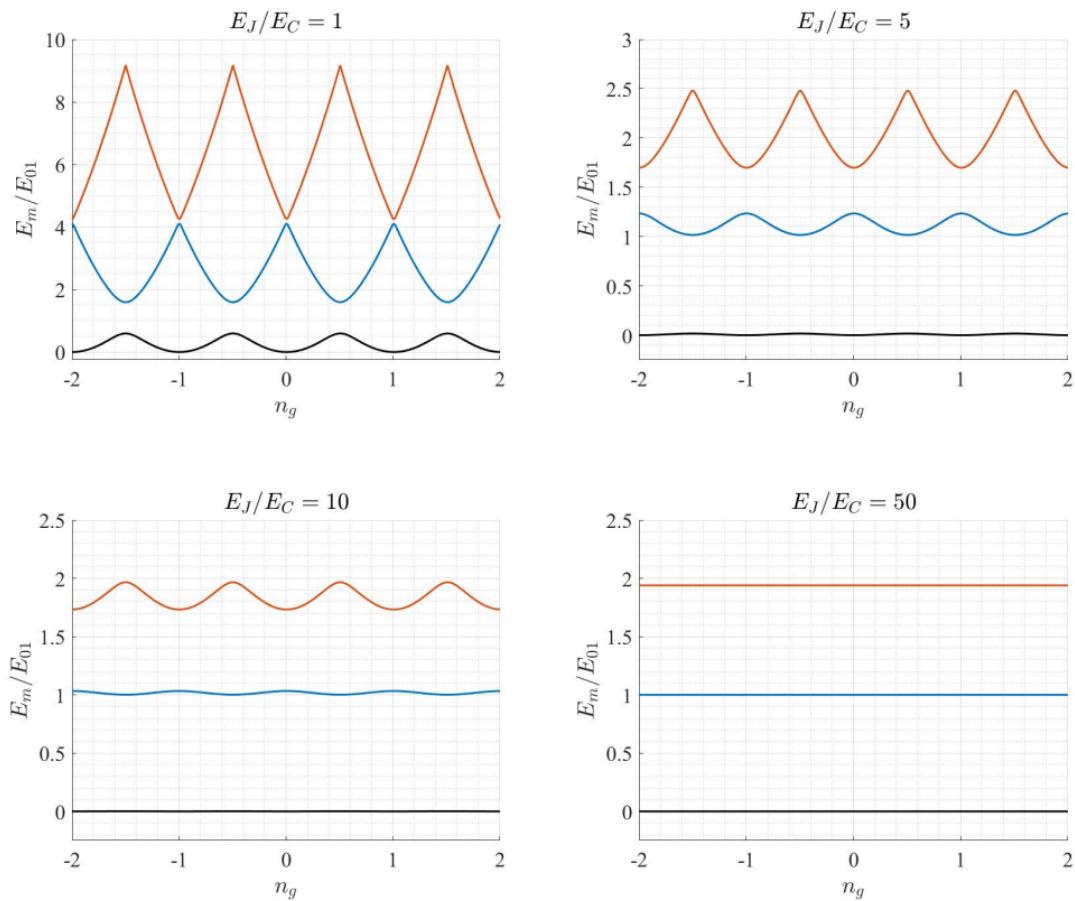


Figure 1.4: Energy levels E_m of the hamiltonian as a function of n_g normalized with respect to E_{01} for different values of the ratio E_J/E_C .

Credits: [11]

1.2.3 Transmon qubit

One of the main drawbacks of the Cooper Pair Box (CPB) qubit, which ultimately led to its replacement by other qubit architectures, is its limited coherence time. The transmon qubit was introduced specifically to address this issue, with the goal of improving the dephasing time of the CPB. The key idea behind the transmon is to reduce the sensitivity of the energy levels to fluctuations in the gate charge—effectively flattening the energy bands—by increasing the ratio between the Josephson energy $E_J/E_C \gg 1$, this architecture was first proposed in [9], the first and more straightforward method to increase this ratio is to enlarge the capacitance of the qubit, which reduces the charging energy E_C .

Since the CPB and the transmon qubit have the same electrical circuit they are also described by the same Hamiltonian (1.19). The difference is that in this case the transmon satisfies the condition $E_J/E_C \gg 1$ it is possible to expand the cosine term in (1.19) with a Taylor series and neglect the higher order terms:

$$\hat{H} \approx 4E_C\hat{n}^2 + \frac{1}{2}E_J\hat{\phi}^2 - \frac{E_J}{4!}\hat{\phi}^4, \quad (1.22)$$

where the last term, proportional to $\hat{\phi}^4$, makes the potential of the transmon slightly anharmonic.

As happens in the standard harmonic oscillator case, the operators $\hat{\phi}$ and \hat{n} satisfy the canonical commutation relation $[\hat{\phi}, \hat{n}] = i\mathbb{I}$, it is possible to introduce the raising and lowering operators \hat{b}, \hat{b}^\dagger as

$$\hat{\phi} = \sqrt{\xi}(\hat{b} + \hat{b}^\dagger), \quad \hat{n} = -\frac{i}{2\sqrt{\xi}}(\hat{b} - \hat{b}^\dagger), \quad (1.23)$$

where $\xi = \sqrt{2E_C/E_J}$.

Substituting equations (1.23) in the Hamiltonian, equation (1.22) becomes

$$\hat{H} = \sqrt{8E_JE_C}\hat{b}^\dagger\hat{b} - \frac{E_C}{12}(\hat{b} + \hat{b}^\dagger)^4. \quad (1.24)$$

Given equation (1.24) it is possible to solve the eigenvalue problem $\hat{H}|k\rangle = E_k|k\rangle$ and calculate the energy levels E_k . The first term of Hamiltonian (1.24) is the harmonic oscillator contribution with eigenstates $|k\rangle$ and eigenvalues $\sqrt{8E_JE_C}k$. Since $E_C \ll E_J$, the second term $\hat{V} = -\frac{E_C}{12}(\hat{b} + \hat{b}^\dagger)^4$ represents a small perturbative contribution to the Hamiltonian and can be treated using perturbation theory. The first-order correction to the energy levels is given by the diagonal matrix elements of the perturbation operator: $\Delta E_k^{(1)} = \langle k|\hat{V}|k\rangle$. It is possible to verify that $\langle k|\hat{V}|k\rangle = -\frac{E_C}{12}(6k^2 + 6k + 3)$. Thus the eigenenergies of the transmon Hamiltonian are

$$E_k \approx \sqrt{8E_JE_C}k - \frac{E_C}{2}(k^2 + k). \quad (1.25)$$

As mentioned before, the qubit frequency is defined as $f_q = f_{01} = (E_1 - E_0)/\hbar$ which yields

$$f_{01} \approx (\sqrt{8E_JE_C} - E_C)/\hbar \quad (1.26)$$

As explained at the beginning of this section, a large E_J/E_C ratio makes the transmon qubit significantly less sensitive to charge noise. However, this improvement comes at the expense of reduced anharmonicity in the energy level spectrum. The anharmonicity η is defined as the difference between the second and first transition energies, relative to the first transition energy:

$$\eta = \frac{(E_2 - E_1) - (E_1 - E_0)}{\hbar} = \omega_{12} - \omega_{01}. \quad (1.27)$$

For a transmon, the anharmonicity η is negative, reflecting the fact that the level spacing decreases with increasing energy. Ideally, the absolute value $|\eta|$ should be sufficiently large to allow external microwave drives to selectively address the $|0\rangle \leftrightarrow |1\rangle$ transition without inadvertently exciting higher-energy states.

1.2.4 Flux-tunable transmon

To implement certain two-qubit gate schemes, such as swap interactions, it is essential to tune the qubit frequency. A common approach to achieving this is by adding an extra junction to the transmon, the most common configuration is the SQUID (Superconducting QUantum Interference Device). In the SQUID configuration two Josephson junctions are connected in parallel on a superconducting loop, as shown in Figure (1.5).

Starting from the Hamiltonian of the single Josephson junction it is possible to write the Hamiltonian of a SQUID:

$$\hat{H} = 4E_C \hat{n}^2 - E_{J1} \cos \hat{\phi}_1 - E_{J2} \cos \hat{\phi}_2, \quad (1.28)$$

where E_{J1} and E_{J2} are the Josephson energies of the two junctions, and the operators $\hat{\phi}_1$ and $\hat{\phi}_2$ are the phase differences across the junctions.

Because of the quantization of the magnetic flux through the SQUID, the quantities $\hat{\phi}_1$ and $\hat{\phi}_2$ are not independent. In particular, as shown in [7], the difference between $\hat{\phi}_1$ and $\hat{\phi}_2$ follows the following relation:

$$\hat{\phi}_1 - \hat{\phi}_2 = \frac{2\pi}{\Phi_0} \Phi_{\text{ext}} \pmod{2\pi}, \quad (1.29)$$

where Φ_{ext} is the flux of external magnetic field defined as the integral of the magnetic field over the SQUID area.

Equation (1.29) can be simplified and rewritten as

$$\hat{H} = 4E_C \hat{n}^2 - E_J(\Phi_{\text{ext}}) \cos \hat{\varphi}, \quad (1.30)$$

where $\hat{\varphi} = \frac{\hat{\phi}_1 + \hat{\phi}_2}{2}$, and the Josephson energy is flux-dependent:

$$E_J(\Phi_{\text{ext}}) = (E_{J1} + E_{J2}) \left| \cos \left(\pi \frac{\Phi_{\text{ext}}}{\Phi_0} \right) \right| \sqrt{1 + d^2 \tan^2 \left(\pi \frac{\Phi_{\text{ext}}}{\Phi_0} \right)}, \quad (1.31)$$

where $d = \frac{E_{J1} - E_{J2}}{E_{J1} + E_{J2}}$ is the relative junction asymmetry.

Then, it's easy to see that the frequency f_q of a two-junction transmon depends on the magnetic flux $\Phi_q(t)$ through the SQUID loop, the relation between the magnetic flux and qubit frequency is given by [12]

$$f_q(\Phi) = \left(f_q^{\max} + \frac{E_C}{h} \right) \sqrt{d^2 + (1 - d^2) \cos^2 \left(\pi \frac{\Phi}{\Phi_0} \right)} - \frac{E_C}{h}, \quad (1.32)$$

where $f_q^{\max} = (\sqrt{8E_C E_J} - E_C)/h$ is the maximum qubit frequency. For symmetric junctions this relation can be approximated by [13]

$$f_q(\Phi_q) \approx \frac{1}{h} \left(\sqrt{8E_J E_C \cos \left(\pi \frac{\Phi_q}{\Phi_0} \right)} - E_C \right). \quad (1.33)$$

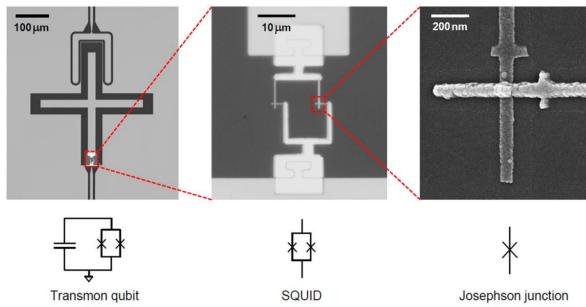


Figure 1.5: Images of a flux tunable transmon qubit. From left to right: the flux tunable transmon qubit, consisting of a large cross-shaped capacitance in parallel with a SQUID to ground, and its corresponding circuit. A zoom in of the SQUID (center), a single Josephson junction (right). Source: [11]

The main drawback of having the qubit frequency depending on the magnetic flux is that the qubit frequency becomes also sensitive to spurious magnetic currents. A strategy to minimize the sensitivity of the transition frequency consists in using a special bias value, known as sweetspot. This corresponds to the charge-degeneracy point, $n_g = 1/2$, where the qubit reaches its maximum transition frequency. At this point, small fluctuations in the bias, either increases or decreases, lead to a small change in frequency, making the qubit less susceptible to noise. Most importantly, since the charge dispersion has no slope there, linear noise contributions cannot change the qubit transition frequency.

1.3 Qubit readout

Up to this point only the physical structure of a transmon qubit was discussed. However, in order to perform quantum computing, it is essential to be able to control and measure its quantum state. One approach to achieve this is to capacitively couple the qubit to both a drive line and a readout line. The drive line is used to manipulate the state of the qubit, while the readout line is employed to measure it. An example of such a system is shown in Figure (1.6), along with the corresponding circuit diagram.

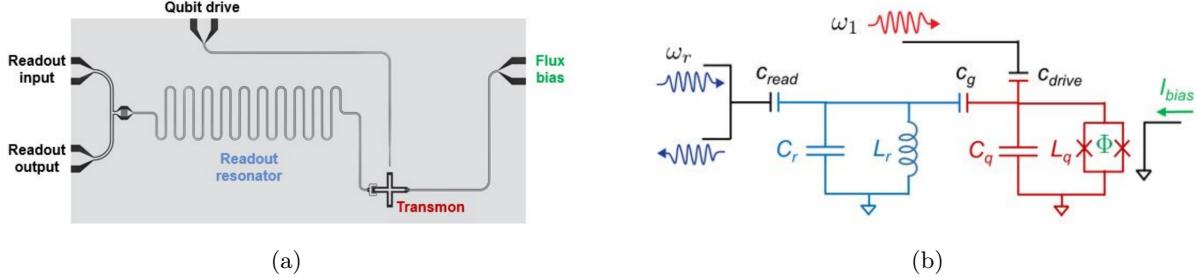


Figure 1.6: Figure (1.6a) shows an example of a single transmon device. Figure (1.6b) shows the equivalent lumped element circuit model of the device in (1.6a), in blue is represented the resonator circuit while in red the transmon qubit circuit. Source: [11]

As mentioned in the introduction of this section, in circuit quantum electrodynamics (cQED), the qubit state is measured via a dispersive interaction between a qubit and a far-detuned microwave resonator.

In Figure (1.6b), in blue is represented the resonator circuit capacitively coupled to the (red) qubit that is used for the readout of the qubit state. The resonator circuit is characterized by a an inductance L_r and a capacitance C_r , then the characteristic frequency is $\omega_r = 1/\sqrt{L_r C_r}$.

The classical Hamiltonian of the resonator can be written as

$$H_r = \frac{Q^2}{2C_r} + \frac{1}{2}C\omega_r^2\Phi^2, \quad (1.34)$$

where Φ is the generalized flux, defined as the time integral of the voltage across the capacitor:

$$\Phi(t) = \int_{-\infty}^t V(t')dt'. \quad (1.35)$$

The quantization of the Hamiltonian involves replacing the classical conjugate variables with their corresponding hermitian operators $\hat{\Phi}$ and \hat{Q} with $[\hat{\Phi}, \hat{Q}] = i\hbar\mathbb{I}$ which leads to

$$\hat{H}_r = \frac{\hat{Q}^2}{2C_r} + \frac{1}{2}C\omega_r^2\hat{\Phi}^2. \quad (1.36)$$

To find the eigenvalues ans eigenvectors of \hat{H}_r it is often convenient to introduce the raising and lowering operators, \hat{a} and \hat{a}^\dagger which satisfy $[\hat{a}, \hat{a}^\dagger] = \mathbb{I}$. These two non-hermitian operators are defined as

$$\hat{\Phi} = \Phi_{\text{zpf}}(\hat{a} + \hat{a}^\dagger), \quad (1.37)$$

$$\hat{Q} = -iQ_{\text{zpf}}(\hat{a} - \hat{a}^\dagger), \quad (1.38)$$

where $\Phi_{\text{zpf}} = \sqrt{\frac{\hbar}{2C_r\omega_r}}$ and $Q_{\text{zpf}} = \sqrt{\frac{C_r\hbar\omega_r}{2}}$ are the zero-point fluctuations. It is then possible to write the Hamiltonian of a microwave resonator in a quantum regime:

$$\hat{H}_r = \hbar\omega_r \left(\hat{a}^\dagger \hat{a} + \frac{1}{2} \right), \quad (1.39)$$

The system of the capacitively coupled transmon and resonator is built in such a way that there is a maximum coupling between the qubit and the resonator. The Hamiltonian of the system reads

$$\hat{H} = \hbar\omega_r \hat{a}^\dagger \hat{a} + 4E_C(\hat{n} + \frac{C_{\text{read}}\hat{V}}{2e})^2 - E_J \cos \hat{\phi} \quad (1.40)$$

$$= \hbar\omega_r \hat{a}^\dagger \hat{a} + 4E_C \hat{n}^2 - E_J \cos \hat{\phi} + \frac{4E_C}{e} \hat{n}_C \hat{V}, \quad (1.41)$$

where $\hat{V} = \hat{Q}/C_r$ is the voltage across the resonator capacitor. Using equations (1.23) and (1.39) the Hamiltonian becomes

$$\hat{H} = \hbar\omega_r \hat{a}^\dagger \hat{a} + \sqrt{8E_J E_C} \hat{b}^\dagger \hat{b} - \frac{E_C}{12} (\hat{b} + \hat{b}^\dagger)^4 + \hbar g(\hat{b}^\dagger - \hat{b})(\hat{a} - \hat{a}^\dagger) \quad (1.42)$$

where the parameter g was introduced. This parameter, known as coupling strength quantifies the strength of the coupling between the qubit and the resonator, it is defined as:

$$g = \frac{2E_C}{\hbar e} \frac{C_{\text{read}}}{C_r} Q_{\text{zpf}} \sqrt{\xi} = \frac{E_C}{\hbar e} \left(\frac{E_J}{2E_C} \right)^{1/4} \frac{C_{\text{read}}}{C_r} \sqrt{2\hbar\omega_r C_r}. \quad (1.43)$$

The coupling strength can be adjusted by varying the capacitance coupling the qubit and the resonator.

When $g \ll \omega_r$ and $g \ll \omega_q$ it is possible to use the rotating wave approximation (RWA) and write the Hamiltonian in the form

$$\hat{H} = \hbar\omega_r \hat{a}^\dagger \hat{a} + \sqrt{8E_J E_C} \hat{b}^\dagger \hat{b} - \frac{E_C}{12} (\hat{b} + \hat{b}^\dagger)^4 + \hbar g(\hat{b}^\dagger \hat{a} + \hat{b} \hat{a}^\dagger). \quad (1.44)$$

Focusing on the first two levels of the transmon we obtain the Jaynes-Cummings Hamiltonian that reads

$$\hat{H} = \hbar\omega_r \hat{a}^\dagger \hat{a} - \frac{\hbar\omega_{01}}{2} \hat{\sigma}_z + \hbar g(\hat{\sigma}_+ \hat{a} + \hat{\sigma}_- \hat{a}^\dagger), \quad (1.45)$$

and describes the interaction between an atom, in this case an artificial atom, with an electromagnetic field in the approximation of the two-level system.

When the coupling strength g is much smaller than the detuning between the qubit and the resonator $\Delta = \omega_q - \omega_r$ the system operates in the dispersive regime. The Jaynes-Cummings Hamiltonian in the dispersive regime, which is the condition in which we operate to perform the qubit readout, can be approximated as

$$\hat{H}_{\text{disp}} = \hbar(\omega_r - \chi \hat{\sigma}_z) \hat{a}^\dagger \hat{a} - \frac{\hbar}{2} (\omega_{01} + \chi) \hat{\sigma}_z, \quad (1.46)$$

where χ is the dispersive shift defined as

$$\chi = \frac{g^2}{\Delta}. \quad (1.47)$$

Equation (1.46) shows that there is a shift in the resonator frequency from ω_r to $\omega_r - \chi$ if the qubit is in the ground state and a shift from ω_r to $\omega_r + \chi$ if the qubit is in the excited state.

The dispersive shift equation (1.47) was derived assuming that the qubit can be approximated as a two level system. Considering also the higher energy levels of the qubit a more accurate expression of the dispersive shift is given by

$$\chi = \frac{g^2}{\Delta(1 + \Delta/\eta)}, \quad (1.48)$$

where η is the qubit anharmonicity.

1.4 Qubit control

Another necessary element for performing quantum computation is the implementation, starting with single-qubit gates. A qubit can be driven into any arbitrary superposition state by applying an electrical pulse with a carefully controlled amplitude, duration, and phase. This pulse is generated by an AC voltage source located outside the dilution refrigerator that hosts the qubit. The driving pulse is brought to the qubit by an on-chip waveguide which is capacitively coupled to the qubit as shown in Figure (1.6b), where the coupling capacitance is indicated with c . This signal path is commonly referred to as a control line or an XY line. The pulse that arrives at the device has the analytical form

$$V_d(t) = A\varepsilon(t) \sin(\omega_d t + \alpha), \quad (1.49)$$

where A is the pulse amplitude in volts, ω_d is the drive frequency in rad/s, α is the phase of the pulse and $\varepsilon(t)$ is the modulation of the pulse; the maximum of $\varepsilon(t)$ is fixed at one. As a first approximation, the envelope of the drive pulse is often chosen to have a Gaussian shape, which is preferred over a square pulse due to its smaller frequency bandwidth, minimizing the excitation of higher energy levels. However,

the study of pulse shapes that minimize leakage to states outside the computational basis remains an active area of research [14].

In a similar way to what was previously done to study the capacitive coupling between the qubit and the resonator, it is possible to derive the hamiltonian of the transmon capacitively coupled to the control line starting from the circuit shown in Figure (1.6b)

$$\hat{H} = 4E_C \left(\hat{n} + \frac{C_d V_d(t)}{2e} \right)^2 - E_J \cos \phi, \quad (1.50)$$

where $E_C = e^2/2C_\Sigma$ and $C_\Sigma = C_d + C_q$. By expanding the parenthesis and dropping a constant term the Hamiltonian can be re-written as

$$\hat{H} = 4E_C \hat{n}^2 - E_J \cos \phi + 2e \frac{C_d}{C_\Sigma} V_d(t) \hat{n}. \quad (1.51)$$

Since $\hat{n} = -i(\hat{b} - \hat{b}^\dagger)/2\sqrt{\xi}$ (from equation (1.23)), the last term of the Hamiltonian can be rewritten:

$$\hat{H}_d = -2e \frac{C_d}{C_\Sigma} V_d(t) \hat{n} \quad (1.52)$$

$$= -i \frac{e}{\sqrt{\xi}} \frac{C_d}{C_\Sigma} V_d(t) (\hat{b} - \hat{b}^\dagger) \quad (1.53)$$

$$= -i \frac{e}{\sqrt{\xi}} \frac{C_d}{C_\Sigma} V_d(t) (|0\rangle \langle 1| - |1\rangle \langle 0|) \quad (1.54)$$

$$= \frac{e}{\sqrt{\xi}} \frac{C_d}{C_\Sigma} V_d(t) \hat{\sigma}_y, \quad (1.55)$$

Regarding the first part of equation (1.51), by focusing on the first two levels of the transmon the qubit Hamiltonian can be rewritten as $\hat{H}_q = -\frac{1}{2}\hbar\omega_q \hat{\sigma}_z$. Then, by substituting the qubit and drive Hamiltonians (equation (1.52)) and the analytical form of $V_d(t)$ in equation (1.51) we obtain

$$\hat{H} = \hat{H}_q + \hat{H}_d = -\frac{\hbar\omega_q}{2} \hat{\sigma}_z + \frac{e}{\sqrt{\xi}} \frac{C_d}{C_\Sigma} A \epsilon(t) \sin(\omega_d t + \alpha) \hat{\sigma}_y \quad (1.56)$$

$$= -\frac{\hbar\omega_q}{2} \hat{\sigma}_z + \hbar\Omega \epsilon(t) \sin(\omega_d t + \alpha) \hat{\sigma}_y, \quad (1.57)$$

$$(1.58)$$

where Ω is the Rabi frequency,

$$\Omega = \frac{e}{\hbar\sqrt{\xi}} \frac{C_d}{C_\Sigma} A. \quad (1.59)$$

The Rabi frequency quantifies the coupling between the control line and the qubit.

To study the dynamics of the system it is convenient to use the rotating frame, the Hamiltonian becomes:

$$\hat{H}' = -\frac{\hbar(\omega_q - \omega_d)}{2} \hat{\sigma}_z + \hbar\Omega \epsilon(t) \sin(\omega_d t + \alpha) (\hat{\sigma}_x \sin \omega_d t + \hat{\sigma}_y \cos \omega_d t) \quad (1.60)$$

$$= -\frac{\hbar(\omega_q - \omega_d)}{2} \hat{\sigma}_z + \hbar\Omega \epsilon(t) \left(\frac{\hat{\sigma}_x}{2} (-\cos(2\omega_d t + \alpha) + \cos \alpha) + \frac{\hat{\sigma}_y}{2} (\sin(2\omega_d t + \alpha) + \sin \alpha) \right) \quad (1.61)$$

$$= -\frac{\hbar(\omega_q - \omega_d)}{2} \hat{\sigma}_z + \frac{\hbar\Omega}{2} \epsilon(t) (\hat{\sigma}_x \cos \alpha + \hat{\sigma}_y \sin \alpha), \quad (1.62)$$

where in the last step I used the rotating wave approximation, meaning that as $\omega_d \approx \omega_q$ the terms $\cos(2\omega_d t + \alpha)$ and $\sin(2\omega_d t + \alpha)$ oscillate rapidly and their contribution to the dynamics can be neglected.

For example, if $\omega_d = \omega_q$ and $\alpha = \frac{\pi}{2}$, the state evolution is described as follows:

$$|\psi\rangle = \exp \left\{ \frac{i}{\hbar} \int_0^{+\infty} \hat{H}(t') dt' \right\} |0\rangle = \exp \left\{ -i \frac{\Omega}{2} \int_0^{+\infty} \epsilon(t') dt' \right\} |0\rangle = e^{-i \frac{\theta}{2} \hat{\sigma}_y} |0\rangle, \quad (1.63)$$

where

$$\theta = \Omega \int_0^{+\infty} \epsilon(t') dt', \quad (1.64)$$

In general, a microwave pulse given by $V_d = A \epsilon(t) \sin(\omega_d t + \alpha)$ implements a single-qubit rotation $R_{\hat{n}(\alpha)}(\theta)$ around an axis \hat{n} that lies on the equator of the Bloch sphere, so that

$$R_{\hat{n}(\alpha)}(\theta) = e^{-\frac{i}{2} \hat{n}(\alpha) \cdot \vec{\sigma} \theta} = e^{-\frac{i}{2} (\hat{\sigma}_x \cos \alpha + \hat{\sigma}_y \sin \alpha) \theta}, \quad (1.65)$$

1.5 Qubit state degradation

The ability to apply quantum gates to qubits is essential for executing complex circuits and operations. However, to ensure their effective implementation, it is crucial to study the qubit decoherence time, as it directly impacts the reliability and performance of quantum computations.

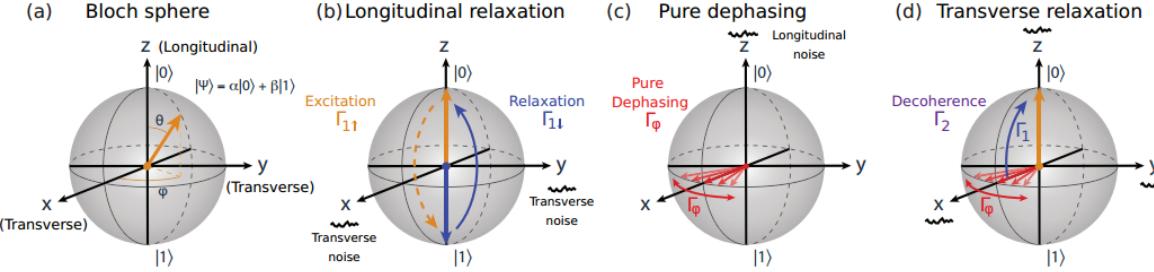


Figure 1.7: Transverse and longitudinal noise represented on the Bloch sphere. (a) Bloch sphere representation of the quantum state $|\psi\rangle = \alpha|0\rangle + \beta|1\rangle$, the z -axis is longitudinal in the qubit frame, while the $x - y$ plane is transverse in the qubit frame. (b) Longitudinal relaxation results from energy exchange between the qubit and its environment, due to transverse noise. (c) Pure dephasing in the transverse plane arises from longitudinal noise along the z axis that fluctuates the qubit frequency. A Bloch vector along the x -axis will diffuse clockwise or counterclockwise around the equator due to the stochastic frequency fluctuations. (d) Transverse relaxation results in a loss of coherence due to a combination of energy relaxation and pure dephasing.

Credits: [15]

1.5.1 Qubit decoherence

By definition, decoherence refers to the loss of coherence in a quantum system, meaning that the relative phase between quantum states becomes randomized due to interactions with the environment. Decoherence is typically characterized by two key time constants: the energy relaxation time T_1 and the dephasing time T_2 . The energy relaxation time describes the characteristic time over which a qubit in an excited state decays to its ground state due to interactions with the environment. The decay follows an exponential law:

$$p_e(t) = p_e(0)e^{-\Gamma_1 t}, \quad (1.66)$$

where $p_e(t)$ is the probability of the qubit being in the excited state at time t , $p_e(0)$ the probability of the qubit being in the excited state at time $t = 0$ and $\Gamma_1 = 1/T_1$ is the decay rate of the qubit from state $|0\rangle$ to state $|1\rangle$. Energy relaxation is primarily due to transverse noise, namely noise that couples the qubit through $\hat{\sigma}_x$ and/or $\hat{\sigma}_y$, it is caused by coupling to electromagnetic noise and other dissipative processes that induce energy exchange with the environment.

Longitudinal noise instead is due to environmental fluctuations coupling to $\hat{\sigma}_z$ that randomize the phase of the superposition state without causing energy exchange with the environment. The characteristic dephasing time of a qubit è indicato con $T_2^* = 1/\Gamma_\phi$ dove Γ_ϕ is the pure dephasing rate.

The transverse relaxation rate Γ_2 describes the combined effects of pure dephasing and energy relaxation and is defined as

$$\Gamma_2 = \frac{\Gamma_1}{2} + \Gamma_\phi, \quad (1.67)$$

or equivalently

$$\frac{1}{T_2} = \frac{1}{2T_1} + \frac{1}{T_2^*}. \quad (1.68)$$

T_2 is known as decoherence time and quantifies how quickly a quantum superposition state loses phase coherence due to environmental fluctuations. For this reason, in an ideal system where pure dephasing is absent T_2 is simply limited by energy relaxation via the relaxation

$$T_{2,\max} = 2T_1. \quad (1.69)$$

This result follows from the Bloch equations [15], where both energy relaxation and phase damping contribute to coherence loss. However, in real systems, additional dephasing mechanisms, such as low-frequency noise from charge or flux fluctuations, reduce T_2 below this upper bound and limits it to (1.68).

1.5.2 Noise models

Quantum noise mentioned in the previous subsection can be described using quantum operations known as quantum channels. A quantum operation is a mathematical transformation that describes how a quantum state changes as a consequence of a physical process. Formally, it is a map \mathcal{E} that transforms a quantum state described by a density operator $\hat{\rho}$ into another state described by a new density operator $\hat{\rho}'$:

$$\mathcal{E}(\hat{\rho}) = \hat{\rho}'. \quad (1.70)$$

Amplitude damping channel

The amplitude damping channel models energy dissipation due to interaction with a zero-temperature environment. The quantum map which describes the process is

$$\mathcal{E}_{\text{ad}}(\hat{\rho}) = \hat{E}_0 \hat{\rho} \hat{E}_0^\dagger + \hat{E}_1 \hat{\rho} \hat{E}_1^\dagger, \quad (1.71)$$

where the Kraus operators:

$$\hat{E}_0 = \begin{bmatrix} 1 & 0 \\ 0 & \sqrt{1-\gamma} \end{bmatrix}, \quad \hat{E}_1 = \begin{bmatrix} 0 & \sqrt{\gamma} \\ 0 & 0 \end{bmatrix}, \quad (1.72)$$

where $p = 1 - e^{-t/T_1}$ represents the probability of relaxation.

Generalized amplitude damping channel

The generalized amplitude damping channel models energy dissipation in a thermal environment where both relaxation and thermal excitation occur.

$$\mathcal{E}_{\text{gad}}(\hat{\rho}) = p \left(\hat{E}_0 \hat{\rho} \hat{E}_0^\dagger + \hat{E}_1 \hat{\rho} \hat{E}_1^\dagger \right) + (1-p) \left(\hat{E}_2 \hat{\rho} \hat{E}_2^\dagger + \hat{E}_3 \hat{\rho} \hat{E}_3^\dagger \right), \quad (1.73)$$

where:

$$\hat{E}_0 = \sqrt{p} \begin{bmatrix} 1 & 0 \\ 0 & \sqrt{1-\gamma} \end{bmatrix}, \quad = \sqrt{p} \begin{bmatrix} 0 & \sqrt{\gamma} \\ 0 & 0 \end{bmatrix}, \quad (1.74)$$

$$\hat{E}_2 = \sqrt{1-p} \begin{bmatrix} \sqrt{1-\gamma} & 0 \\ 0 & 1 \end{bmatrix}, \quad \hat{E}_3 = \sqrt{1-p} \begin{bmatrix} 0 & 0 \\ \sqrt{\gamma} & 0 \end{bmatrix}, \quad (1.75)$$

where γ represents the thermal excitation probability and p describes the system-bath interaction strength.

Phase Damping channel

The phase damping channel describes pure dephasing, where the qubit phase is randomized without energy exchange. The Kraus operators for this channel are:

$$\hat{E}_0 = \begin{bmatrix} 1 & 0 \\ 0 & \sqrt{1-\lambda} \end{bmatrix}, \quad \hat{E}_1 = \begin{bmatrix} 0 & 0 \\ 0 & \sqrt{\lambda} \end{bmatrix}, \quad (1.76)$$

where $\lambda = 1 - e^{-t/T_2^*}$ quantifies the probability of phase randomization.

Depolarizing channel

The depolarizing channel captures the combined effects of relaxation (T_1) and pure dephasing (T_2^*), making it a natural model for total decoherence. The transformation applied by this channel is given by:

$$\hat{\rho} \rightarrow (1-q)\hat{\rho} + \frac{q}{3}(\hat{\sigma}_x \hat{\rho} \hat{\sigma}_x + \hat{\sigma}_y \hat{\rho} \hat{\sigma}_y + \hat{\sigma}_z \hat{\rho} \hat{\sigma}_z), \quad (1.77)$$

meaning that the depolarizing channel leaves $\hat{\rho}$ unchanged with probability $(1-q)$, while with probability $q/3$ one of the Pauli operators is applied to it. The corresponding Kraus operators are:

$$\hat{E}_0 = \sqrt{1-q}\mathbb{I}, \quad \hat{E}_1 = \sqrt{\frac{q}{3}}\hat{\sigma}_x, \quad \hat{E}_2 = \sqrt{\frac{q}{3}}\hat{\sigma}_y, \quad \hat{E}_3 = \sqrt{\frac{q}{3}}\hat{\sigma}_z. \quad (1.78)$$

This channel models both energy relaxation and phase randomization, effectively representing total decoherence.

An alternative representation of the depolarizing channel describes the process in which the system state ρ is replaced by the maximally mixed state $\frac{\mathbb{I}}{2}$ with probability d :

$$\mathcal{E}_{dc}(\rho) = p \frac{\mathbb{I}}{2} + (1 - p)\rho. \quad (1.79)$$

where the following relation holds between p and q , $q = 3p/4$.

Chapter 2

Qubit calibration

In this chapter I will describe the process of calibration for superconducting flux-tunable transmon on the hardware located in the QRC (Quantum Research Center) Laboratory of the TII (Technology & Innovation Institute) in Abu Dhabi.

2.1 Experimental setup

All the results presented in this work were obtained using the Contralto-D chip [16], which offers up to 21 fully connected qubits and 4 isolated qubits, for a total of 25 physical qubits. The distinction between fully connected and isolated qubits is important as only the fully connected subset supports direct two-qubit gate operations, which are essential for implementing entangling gates and complex quantum circuits. Isolated qubits, while still operational for single-qubit tasks, do not participate in multi-qubit interactions and thus are not functionally equivalent in terms of computational capabilities. The topology of the qubit is shown in Figure (2.1b).

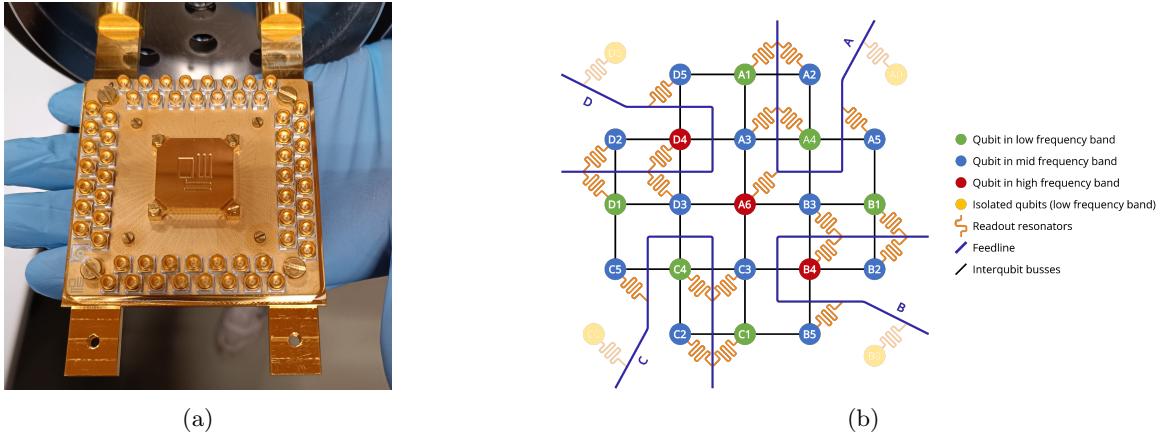


Figure 2.1: Figure (2.1a): Picture of the Contralto-D chip from QuantWare. Figure (2.1b): Topology of the Contralto-D chip from QuantWare.

As discussed in the previous chapter, the behavior of Josephson junctions and SQUIDs relies critically on the superconducting state of the materials involved. To achieve and maintain this regime, it is essential that the superconducting elements operate well below their critical temperature. For this reason, the Contralto-D chip is installed at the lowest temperature stage of the cryostat, where the required thermal conditions for superconductivity are met. This ensures the proper functioning of the quantum hardware and enables the realization of coherent quantum operations.

These systems achieve ultra-low temperatures by exploiting the unique quantum properties of helium-3 (^3He) and helium-4 (^4He) isotopes in a dilution process. At the core of a dilution refrigerator is a mixing chamber, where the cooling mechanism takes place. When a mixture of ^3He and ^4He is cooled below approximately 870 millikelvin, the two isotopes phase-separate into a ^3He -rich phase and a ^3He -dilute phase. The key principle is that when ^3He atoms cross the phase boundary—from the concentrated phase into the dilute phase—they absorb energy from their surroundings. This process is endothermic and is

the fundamental source of cooling in the dilution refrigerator.

The system operates as a closed loop: ^3He gas is circulated using a combination of sorption pumps and still pumps, which remove ^3He vapor from the still (typically at 600–800 mK), recondense it at a higher stage, and reintroduce it into the mixing chamber. The refrigerator includes several thermalization stages—typically at 50 K, 4 K, 800 mK, 100 mK, and finally below 20 mK—each connected to a corresponding cooling stage and separated by radiation shields and thermal filters to minimize heat load and noise from higher-temperature stages. Dilution refrigerators are highly stable and capable of reaching base temperatures below 10 mK, with hold times on the order of days or even weeks. These temperatures are crucial for achieving the low thermal noise and long coherence times necessary for high-fidelity quantum operations in superconducting circuits. Specifically the cryostat employed in the lab is the XLDsl from Bluefors [17], an image of the cryostat is shown in Figure (2.2).

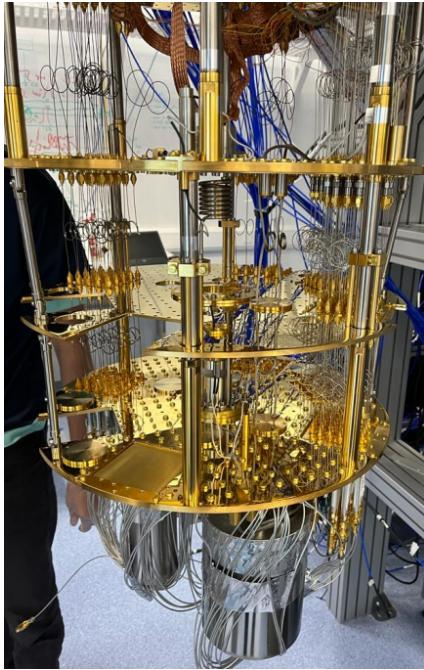


Figure 2.2: Picture of the XLDsl dilution refrigerator at the QRC Lab

Outside the cryostat, the control and readout of superconducting qubits are managed by dedicated room-temperature electronics. These systems are responsible for generating the microwave pulses used to drive single- and two-qubit gates, as well as for acquiring and processing the output signals that encode the qubit states. Typically they include arbitrary waveform generators (AWGs), microwave sources, mixers, digitizers, and field-programmable gate arrays (FPGAs). The generated microwave pulses are shaped and modulated at room temperature before being attenuated and routed to the cryogenic environment. Similarly, signals returning from the qubits are amplified and digitized for state discrimination and further processing. The electronics employed in the lab for the control of the `qw11q` is the OPX1000 platform by Quantum Machines [18].

The software I used for the calibration of the qubits and the subsequent experiments is **Qibocal** ([19], [20], [21]), while the backend for communication with the laboratory instruments is **Qibolab** ([22], [23], [24]). **Qibolab** is the control layer responsible for managing and executing low-level instructions on the hardware, bridging high-level quantum models and physical quantum platforms. It is designed to support diverse experimental setups and allows the researcher to define custom hardware configurations through a platform abstraction and to execute custom pulse sequences using both commercial and open-source firmware. The communication between **Qibolab** and the quantum hardware is structured and modular, relying on a stack that includes instrument drivers, pulse control logic, and a compiler that translates abstract quantum gates into hardware-specific instructions. This structure enables compatibility with heterogeneous platforms and facilitates the development of experimental drivers tailored to different laboratory environments. **Qibocal** interfaces directly with **Qibolab** to apply calibration protocols on the physical device. The routines deployment takes place through the interpretation of declarative runcards written in YAML. **Qibocal** allows an easy execution of pulse sequences, collection of measurement data, and interpretation of the results through the reports that are automatically generated upon completion

of the routine.

2.2 Single qubit calibration experiments

The first task that I needed to complete at the beginning of my thesis work was the calibration of at least a line of the superconducting qubits of the Contralto-D chip using the `Qibocal` library. From this point onward, for the sake of brevity, I will refer to the chip interchangeably as Contralto-D or `qw11q`, which is the name of the node under which it is registered on the QRC computing cluster. In the following I will describe the experiments that I performed and commenting on the results.

2.2.1 Resonator calibration

Before starting with the calibration of the gates necessary for quantum computing it is necessary to characterize the qubit and calibrate the readout pulses. For this reason the calibration process starts with the characterization of the resonator coupled to the qubit that will be used to perform non-destructive measurements of the qubit state.

Resonator spectroscopy

The first step to calibrate the readout pulse is to characterize the resonator is to find the resonator frequency, that is the transition frequency for the resonator. At this frequency, a distinct difference in the transmitted signal can be observed depending on the type of resonator used. In the case of a 3D cavity resonator, the signal appears amplified, whereas for a 2D planar resonator, the signal tends to be more strongly absorbed. Regardless of the resonator type, the response typically exhibits a Lorentzian-shaped peak: this peak is positive for 3D cavities due to the amplification effect, and negative for 2D resonators due to their greater absorption.

The outcome of this experiment is strongly influenced by the amplitude of the excitation pulse. To reliably determine the resonator frequency, the pulse duration can be fixed on the order of microseconds, which is sufficient to observe the relevant signal response. However, selecting an appropriate amplitude requires more careful consideration. When the amplitude is high, the signal becomes more prominent, improving the signal-to-noise ratio (SNR) and making it easier to identify the resonator's response. If the amplitude is increased too much, however, it can drive the system out of the superconducting regime. In this case, the resonator becomes effectively decoupled from the qubit, and the frequency observed corresponds to the so-called bare resonator frequency. During the first calibration operations it is not necessary to have the qubit coupled to the resonator but is helpful to improve the SNR to have a more defined peak in the signal, for this reason the first calibration routines are run in a high-power regime.

In this experiment the main variables on which the experimenter can operate are the frequency range that is willing to scan and the step of the scan. Regarding the frequency range a very wide scan can be useful if nothing is known about the studied resonator, but usually the design parameters are provided by the producer; it is possible that they are not exact, but can give an idea of the region to scan. Regarding the step for the scan, usually a step of 200 MHz can be used to probe the resonator frequency but can be reduced if a more precise scan is needed. An example of measurement of the bare resonator frequency is shown in Figure (2.3).

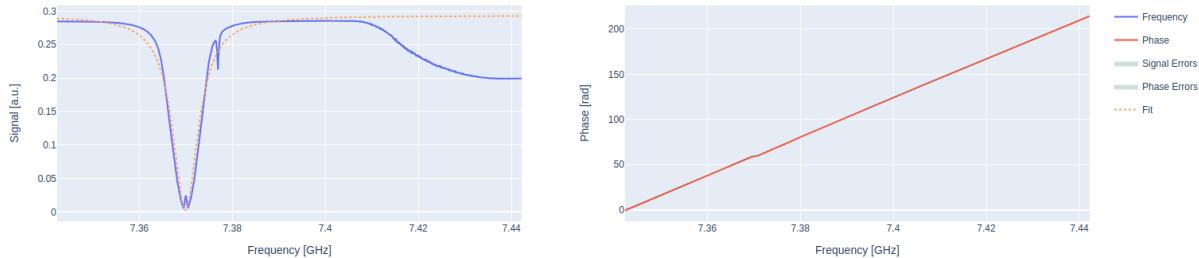


Figure 2.3: Output of resonator spectroscopy with high power on qubit B2.

An example of the measurement of the resonator frequency in the low-power regime is shown in Figure (2.4).

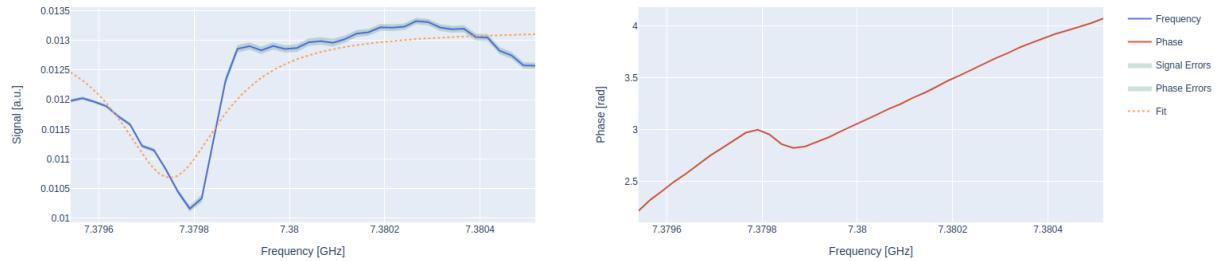


Figure 2.4: Output of resonator spectroscopy with low power on qubit B2.

Resonator punchout

After having found the bare resonator frequency it is necessary to identify the resonator frequency under qubit coupling. To do this it is possible to repeat the spectroscopy, this time over a narrower frequency range and for varying pulse amplitudes. The resonator frequency is expected to depend strongly on amplitude: it remains constant in the high-power regime, shifts during an intermediate transition phase, and stabilizes again at a different value once the qubit-resonator interaction becomes significant. For this calibration protocol the experimenter must carefully choose the frequency and amplitude range to scan to avoid extremely long experiments, at the same time the frequency and amplitude steps must be small enough to clearly identify the frequency and amplitude in which the shift happens.

The expected result from a resonator punchout experiment is shown in Figure (2.5)

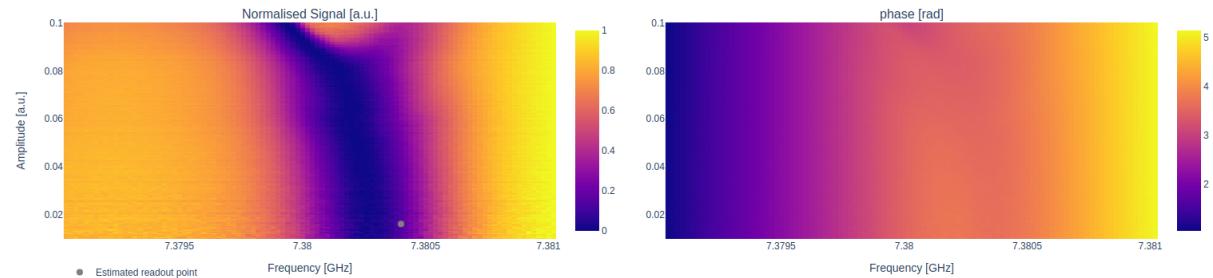


Figure 2.5: Output of resonator punchout on qubit B2.

Resonator flux dependence

As explained in section (1.2.4) it is suggested to work at the qubit sweetspot, where the qubit frequency is less sensitive to magnetic flux fluctuations. For this reason, it is useful to estimate the location of the sweet spot prior to performing qubit spectroscopy, enabling measurements to be carried out near the optimal bias point. This can be accomplished by exploiting the qubit-resonator coupling: since the qubit frequency depends on the magnetic flux, as shown in equation (1.32), and the qubit is coupled to the resonator, the resonator frequency also exhibits a flux dependence. The resonator detuning in the transmon regime ($E_J \gg E_C$) can be computed as

$$f_r(\Phi) = f_r^{\text{bare}} + \frac{g^2 \sqrt[4]{d^2 + (1 - d^2) \cos^2 \left(\pi \frac{\Phi}{\Phi_0} \right)}}{f_r^{\text{bare}} - f_q(\Phi)}, \quad (2.1)$$

where f_r^{bare} is the bare resonator frequency, g^2 is the coupling between the transmon and the resonator, d is the junctions asymmetry, E_C is the charging energy, E_J the Josephson energy and $\Phi_0 = h/2e$ is the flux quanta.

In **Qibocal** it is possible to perform a resonator flux dependence experiment to measure and fit and the curve described by Eq. (2.1). With this routine the experimenter performs a scan of the both the external bias and frequency, this experiment provides an approximate indication of the sweetspot. An example of the output of this experiment is shown in Figure (2.6)

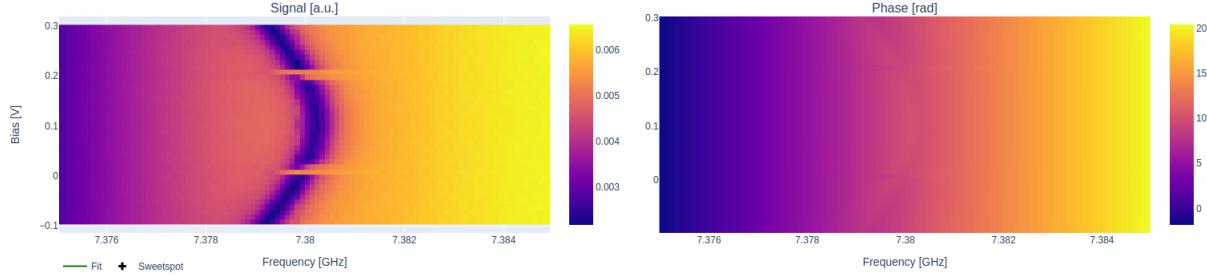


Figure 2.6: Output of resonator flux dependence routine on qubit B2.

2.2.2 Qubit calibration

After having determined all the readout parameters it is possible to continue the calibration process by calibrating the qubit.

Qubit spectroscopy

To determine the resonance frequency of a qubit, a qubit spectroscopy experiment is performed, which, unlike resonator spectroscopy, requires a two-tone approach. While resonator spectroscopy is typically a single-tone measurement used to identify the resonator's response, qubit spectroscopy involves applying a drive tone to the qubit followed by a readout tone to detect the qubit state. This method becomes essential after an initial estimate of the readout frequency and amplitude has been obtained from a resonator punchout experiment. In this protocol, a drive pulse of variable frequency ω is sent through the qubit drive line. If the drive frequency is far detuned from the qubit transition frequency ω_q , it will have no appreciable effect on the qubit state, and the measured signal will remain unchanged. However, as ω approaches ω_{01} the drive pulse can induce transitions between the qubit states. This excitation modifies the qubit population and, consequently, the resonator response, which is sensitive to the qubit state due to their dispersive coupling. When the drive frequency is near resonance and the pulse is sufficiently long, the qubit may reach a maximally mixed state leading to a detectable change in the readout signal amplitude. By sweeping the drive frequency and recording the corresponding readout amplitudes, one can plot a spectroscopy curve that reveals a Lorentzian dip or peak centered at the qubit transition frequency—opposite in direction to the Lorentzian feature observed in the resonator spectroscopy, due to the nature of the state-dependent dispersive shift.

An example of the output of qubit spectroscopy experiment is shown in Figure (2.7)

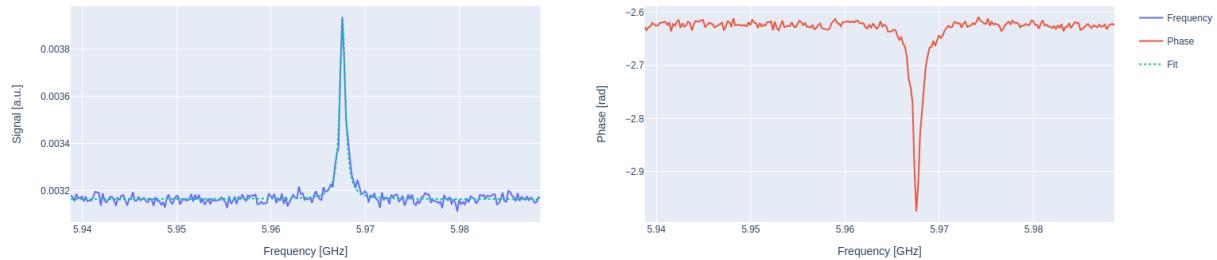


Figure 2.7: Output of qubit spectroscopy on qubit B2.

Qubit EF spectroscopy

Qubit spectroscopy can also be extended to probe transitions to higher excited states beyond the first excited state. Directly observing these higher-level transitions typically requires significantly increased drive power, which may exceed the safe operational limits of the experimental setup. An alternative and more controlled approach involves first preparing the qubit in state $|1\rangle$, followed by a standard spectroscopy sequence to induce the $|1\rangle \leftrightarrow |2\rangle$ transition.

Note that the description of this calibration routine has been included here for the sake of clarity and continuity of exposition; however, since it requires the qubit to be in the $|1\rangle$ state at the beginning of

the spectroscopy, it can only be performed after the execution of a single-shot classification (see Section (2.2.5)).

Qubit flux dependence

By performing a resonator flux dependence the experimenterer obtained a first estimate for the sweetspot, now by performing a qubit flux spectroscopy. In this experiment a constant DC current signal is sent to the qubit through the flux line with voltage and amplitude fixed for the bias level and then a qubit spectroscopy is performed. This procedure is repeated for different bias levels and different drive frequencies. The expected result of the routine is shown in Figure (2.8).

When calibrating the qubit sweetspot it is important to consider that on a superconducting qubatum chip qubits are not completely isolated systems; rather, they interact with one another through both intentional couplings (e.g capacitive or inductive couplers) and unintended cross-talk mechanisms. The mutual influence that qubits exert on each other's affect their effective operating parameters, particularly their flux-dependent transition frequencies. This means that also the location of the sweetspot can be affected by the biasing of neighboring qubits, this occurs because the flux-tuning circuitry is not perfectly isolated; biasing one qubit can induce a spurious flux in nearby qubits, effectively shifting their frequency spectra and hence their sweetspots. Consequently, calibrating each qubit in isolation may yield misleading results, therefore, to accurately determine and operate all qubits at their true sweetspots under realistic experimental conditions, it is important to perform simultaneous calibration. This ensures that all mutual interactions and cross-couplings are properly accounted for, leading to a more stable and predictable multi-qubit operation.

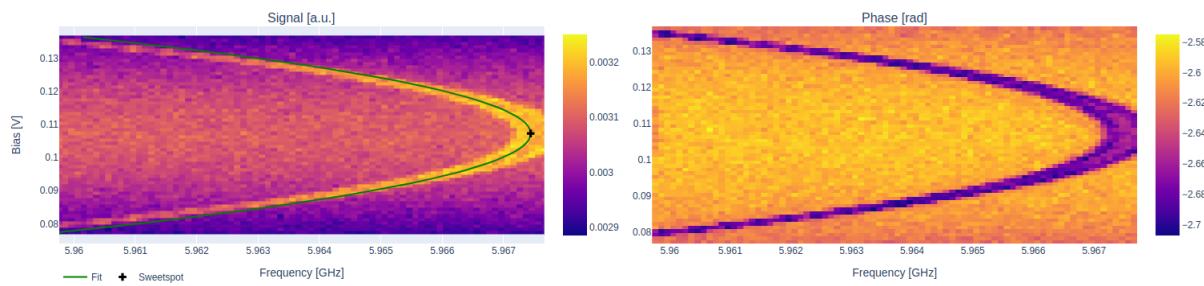


Figure 2.8: Output of the qubit flux dependence on qubit B2.

2.2.3 Drive pulse calibration

After defining all the readout parameters and the parameters for qubit control, it is necessary to calibrate the parameters of the drive pulses to be able to control rotations on the Bloch sphere. With Rabi experiments is possible to calibrate the drive pulses to perform precise rotations (see [25], [26]) within approximately 40 ns. This duration is typical for single-qubit gates in superconducting qubit systems, as it represents a balance between fast control and minimal spectral leakage. Faster gates would require higher drive amplitudes, which can increase errors due to leakage into higher transmon levels or induce unwanted transitions in nearby qubits through cross-resonance or crosstalk effects [9], [27]. On the other hand, significantly longer gates are more susceptible to decoherence, particularly due to T_1 relaxation and T_2 dephasing processes, which limit overall gate fidelity [28].

Usually the first step of calibrating drive pulses consists in calibrating a π -pulse, namely an X -gate. The goal of the Rabi experiment is to tune the amplitude or duration of the drive pulse, in order to excite the qubit from the ground state up to state $|1\rangle$.

Rabi oscillation experiments

To experimentally probe the coherent control of a qubit in **Qibocal** it is possible to perform different variations of the Rabi oscillation experiments, which provides direct evidence of the qubit's ability to undergo controlled rotations under a resonant microwave drive, as described in Section (1.4). In this experiment, a microwave pulse of the form

$$V_d(t) = A\varepsilon(t) \sin(\omega_d t + \alpha), \quad (2.2)$$

is applied through the control line (XY line), and coupled capacitively to the qubit. The qubit is initialized in its ground state $|0\rangle$, and the drive frequency ω_d is set close to the qubit transition frequency ω_q . The envelope $\varepsilon(t)$ is typically Gaussian, and its shape is kept constant throughout the experiment. The key parameter that is varied is the amplitude A of the drive pulse (or alternatively, the duration of the pulse while keeping amplitude constant).

To better understand the evolution of the system, it is possible to consider the driven qubit in the interaction picture. The qubit state can be written as:

$$|\psi(t)\rangle = C_0(t)e^{+i\omega_q t/2}|0\rangle + C_1(t)e^{-i\omega_q t/2}|1\rangle. \quad (2.3)$$

Solving the time-dependent Schrödinger equation for state (2.3) under Hamiltonian(1.58) yields the following expressions:

$$C_0(t) = e^{-i\Delta_d t/2} \left[\cos\left(\frac{\Omega_R t}{2}\right) + i \frac{\Delta_d}{\Omega_R} \sin\left(\frac{\Omega_R t}{2}\right) \right], \quad (2.4)$$

$$C_1(t) = i \frac{\Omega}{\Omega_R} e^{i\Delta_d t/2} \sin\left(\frac{\Omega_R t}{2}\right), \quad (2.5)$$

where $\Delta_d = \omega_q - \omega_d$ is the detuning, and

$$\Omega_R = \sqrt{\Omega^2 + \Delta_d^2}, \quad (2.6)$$

is the generalized Rabi frequency.

The probability of finding the qubit in the excited state at time t is given by:

$$P_1(t) = |C_1(t)|^2 = \frac{\Omega^2}{\Omega^2 + \Delta_d^2} \sin^2\left(\frac{\Omega_R t}{2}\right). \quad (2.7)$$

This result shows that the probability oscillates in time with frequency Ω_R , and the amplitude of the oscillation is reduced when the drive is off-resonant ($\Delta_d \neq 0$). The closer the drive is to resonance, the higher the probability of full excitation.

In the resonant case, when $\omega_d = \omega_q$ and $\Delta_d = 0$ the generalized Rabi frequency reduces to $\Omega_R = \Omega$ and the excited state probability becomes

$$P_e(t) = P_1(t) = \sin^2\left(\frac{\Omega t}{2}\right) = \frac{1}{2}(1 - \cos(\Omega t)), \quad (2.8)$$

which is the curve that is used in the fit when performing a Rabi amplitude experiment.

An example of the output of a Rabi amplitude experiment is shown in figure (2.9).

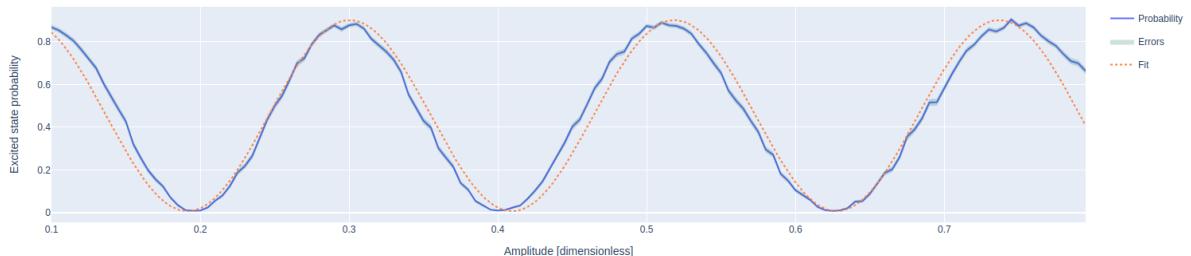


Figure 2.9: Output of the Rabi amplitude routine on qubit B2.

A variation of the Rabi amplitude experiment consists of the Rabi length experiment, where instead of the amplitude, the parameter that is varied is the duration of the drive pulse. The qubit is again initialized in the ground state $|0\rangle$ and subjected to a resonant microwave drive ($\omega_d = \omega_q$). In this case, the pulse envelope $\varepsilon(t)$ and amplitude A are held constant, while the total duration of the pulse is incrementally varied. This enables the observation of coherent oscillations in the excited state population as a function of pulse length. However, unlike the idealized scenario, in realistic conditions the qubit experiences energy relaxation and dephasing, which attenuate the oscillation amplitude over time. To

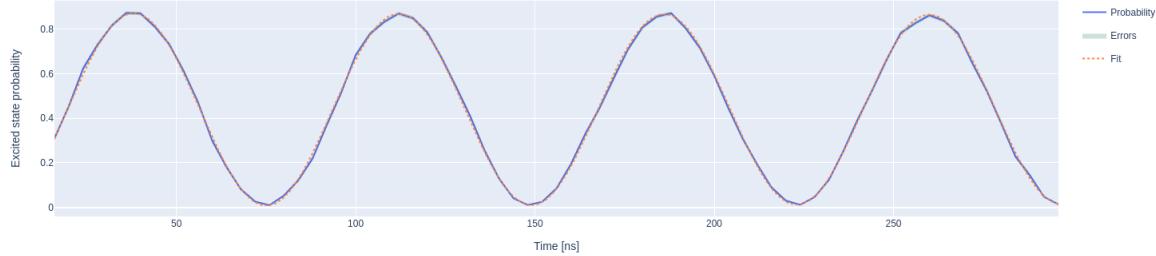


Figure 2.10: Output of the Rabi length routine on qubit B2.

account for these effects, the excited state probability is modeled as represents the effective decay time, incorporating both energy relaxation and dephasing processes.

$$P_e(t) = P_1(t) = \frac{1}{2} \left(1 - e^{-t/\tau} \cos\left(\Omega_R \frac{t}{2}\right) \right), \quad (2.9)$$

where Ω_R is the generalized Rabi frequency and τ

Sometimes, when the qubit frequency is not precisely known, or when systematic effects such as pulse distortions, frequency drifts, or cross-talk may affect the calibration it can be useful to perform a Rabi measurement in which the amplitude and frequency of the drive pulse are scanned. In Qibocal this experiment is implemented as the Rabi amplitude-frequency routine, this experiment provides a comprehensive view of the qubit's response to a range of drive conditions and is particularly useful during initial calibration. This experiment reveals a characteristic chevron-shaped interference pattern in the excitation probability as a function of drive frequency and amplitude. Figure (2.11) shows a possible output of this experiment, in the plot in figure in particular it is possible to see only the first region of the chevron shape with high probability of finding the qubit in an excited state.

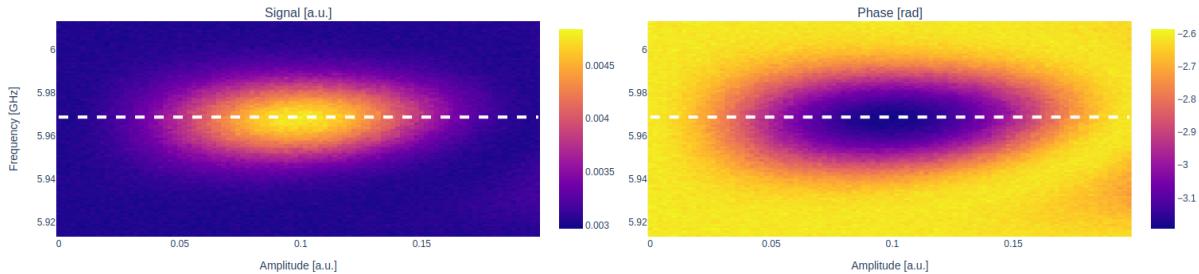


Figure 2.11: Output of the Rabi amplitude-frequency routine on qubit B2.

2.2.4 Fine-tuning calibration

Ramsey experiment

After calibrating the readout and π -pulses, technically should be possible to execute algorithmic experiments on the qubit. However, essential characteristics of the qubit, such as its coherence properties and the precise drive frequency, remain to be determined; without this information, circuit fidelities would be suboptimal. The Ramsey experiment is a simple yet powerful protocol that allows simultaneous investigation of multiple aspects of qubit behavior, including the fine tuning of the drive frequency and verification of coherent control and signal phase consistency. It also enables extraction of the qubit's coherence time T_2^* .

In its standard implementation, the Ramsey sequence consists of two $\frac{\pi}{2}$ -pulses separated by a variable delay τ .

The first $\frac{\pi}{2}$ -pulse brings the qubit from the ground state $|0\rangle$ onto the equatorial plane of the Bloch sphere, placing it in a superposition of $|0\rangle$ and $|1\rangle$. During the delay τ , the qubit state accumulates a relative phase due to its Larmor precession and environmental noise, effectively evolving around the z-axis.

The second $\frac{\pi}{2}$ -pulse projects the final state back onto the measurement basis, and the probability of measuring the excited state $|1\rangle$ is recorded. Repeating this procedure for different delay times τ allows to monitor the decay of coherence in the qubit state.

In the ideal, non-detuned case, where the drive frequency matches the qubit transition frequency, the observed signal exhibits a purely exponential decay toward a baseline value, from which the characteristic decoherence time T_2^* can be extracted. However, small deviations from the ideal $\frac{\pi}{2}$ -pulse, due to imperfect calibration or a mismatch between the drive and qubit frequency lead to more complex behaviour.

Usually a small intentional detuning is applied, in this case the measurement outcomes show a sinusoidal modulation superimposed on the exponential decay so that the resulting curve can be fitted with the equation [29]

$$p_e(\tau) = \frac{1}{2} + \frac{1}{2}e^{-\tau/T_2^*} \cos(\Delta\omega\tau), \quad (2.10)$$

An example of the output of a Ramsey experiment is shown in Figure (2.12)

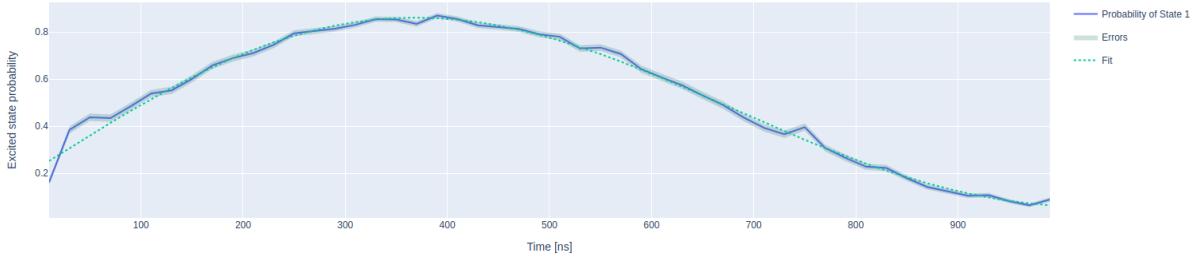


Figure 2.12: Output of the Ramsey amplitude routine on qubit B2.

Flipping experiment

While the Ramsey experiment is typically used to fine-tune the drive frequency of a qubit, the Flipping experiment serves as an effective routine to calibrate the amplitude of the π -pulse, ensuring accurate implementation of $R_x(\pi)$ rotations. This calibration is fundamental for achieving high-fidelity gate operations, and the flipping experiment can often reveal discrepancies in the amplitude that are not evident in Rabi oscillation measurements.

In this protocol, a flip is defined as a pair of consecutive π -pulses, which ideally return the qubit to its initial state due to a full 2π rotation. The experiment begins with the qubit initialized in the ground state $|0\rangle$, followed by an $R_x(\pi/2)$ rotation that places it in an equal superposition state on the equator of the Bloch sphere. Without the initial $R_x(\pi/2)$ rotation, errors in the π -pulses would lead only to a global phase difference, which is not detectable via projective measurements. With the superposition state, however, the qubit's evolution becomes sensitive to amplitude miscalibrations, allowing over- and under-rotations to be distinguished through changes in measurement probability.

Following the initial $\pi/2$ -pulse, the qubit is subjected to a number N of flips chosen by the experimenter. After the flips, a projective measurement is performed. This procedure is repeated for increasing values of N , effectively building a scan of how the qubit state evolves under repeated application of imperfect π -pulses.

In the ideal case, where the π -pulse amplitude is perfectly calibrated, the repeated flips return the qubit to the equatorial plane after each cycle, resulting in a constant measurement probability. The recorded signal should appear as a flat line; however, if the amplitude is miscalibrated, the qubit state will drift on the Bloch sphere, leading to oscillatory behavior in the measured population as a function of the number of flips. This deviation is modeled by a sinusoidal function, from which the amplitude error can be quantitatively extracted. Figure (2.13) shows the result of a flipping routine.

Dispersive shift

A possible strategy to improve the assignment fidelity is to perform a dispersive shift measurement which helps calibrate the readout frequency. To calibrate the readout frequency based on the dispersive shift, two resonator spectroscopy measurements are performed. In the first, the resonator response is measured while the qubit remains in its ground state $|0\rangle$, effectively treating the system as a bare resonator. In the second measurement, a calibrated π -pulse is applied before each acquisition to prepare the qubit in the excited state $|1\rangle$, and the spectroscopy is repeated under otherwise identical conditions. The transmission or reflection spectra from both configurations are then plotted together as a function of probe frequency. At each frequency point, single-shot measurements are collected, and the resulting IQ data are analyzed.

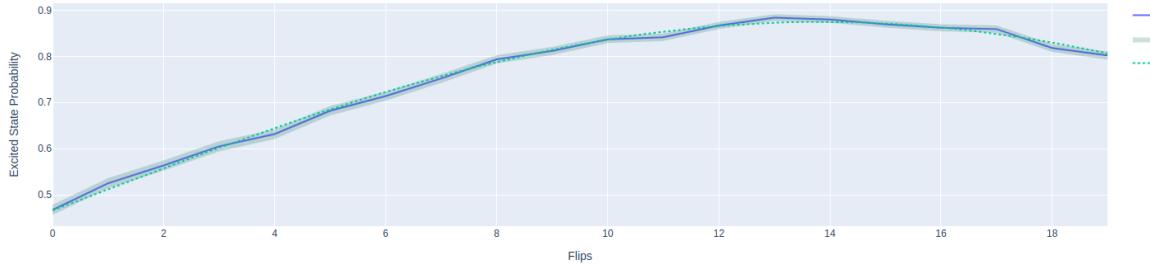


Figure 2.13: Output of the flipping routine on qubit B2.

Specifically, the distance between the centroids of the two resulting IQ distributions, one corresponding to the ground state and the other to the excited state, is computed. The readout frequency that maximizes this centroid separation is selected, as it provides the greatest distinguishability between the two qubit states and thus optimizes the readout contrast.

A typical result for this routine is shown in Figure (2.14)

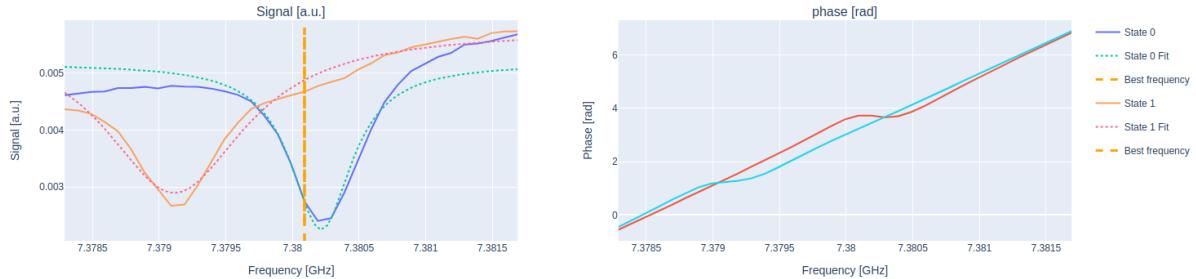


Figure 2.14: Output of the Rabi amplitude routine on qubit B2.

2.2.5 Qubit characterization

T1 & T2 measurement

After having calibrated all the parameters concerning the qubit it is possible to proceed with characterization experiments such as T_1 and T_2 measurements.

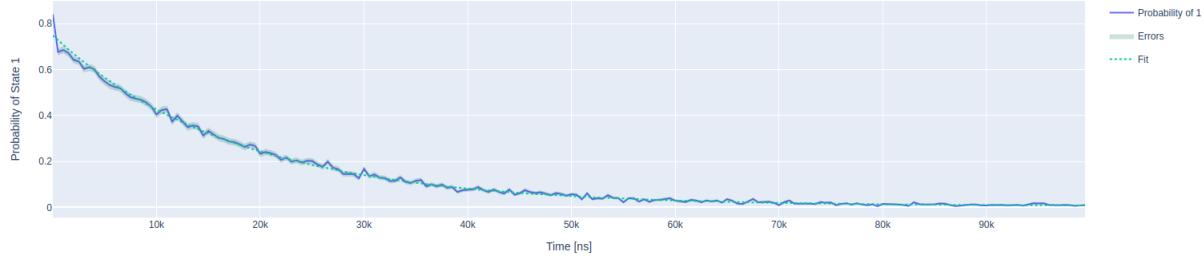
As explained in Section (1.5.1), due to coupling to the environment the qubit in state $|1\rangle$ will decay to the state $|0\rangle$. To measure the characteristic decay time T_1 it is possible to perform a simple experiment where the qubit is initialized in state $|1\rangle$ using a previously calibrated π pulse. After a variable delay $\Delta\tau$, during which the qubit undergoes spontaneous decay due to coupling to the environment, a projective measurement is performed. For $\Delta\tau = 0$ we expect to find the qubit in state $|1\rangle$ while for $\Delta\tau \rightarrow \infty$ the system relaxes fully to the ground state $|0\rangle$. At intermediate times, the probability of finding the qubit in $|1\rangle$ decays exponentially, and the measured excited state population $p_e(t)$ as a function of time t is fitted to the model

$$p_e(t) = A + Be^{-\frac{t}{T_1}}, \quad (2.11)$$

where T_1 is the energy relaxation time and A, B are the fitting parameters. The expected result from this routine is shown in Figure (2.15).

After measuring T_1 it is necessary to measure also the dephasing time of the qubit, T_2 ; to do this in Qibocal, it is possible to perform a T_2 experiment. The acquisition sequence of the experiment is the same as the Ramsey experiment described in Section (2.2.4): a pair of $\pi/2$ -pulses is applied with a variable delay τ in between, during which the qubit freely evolves under a slightly detuned drive. As a result, the measured excited state population exhibits oscillations in time, caused by the accumulation of phase due to the detuning. These oscillations are enveloped by an exponential decay that reflects the qubit's loss of phase coherence over time.

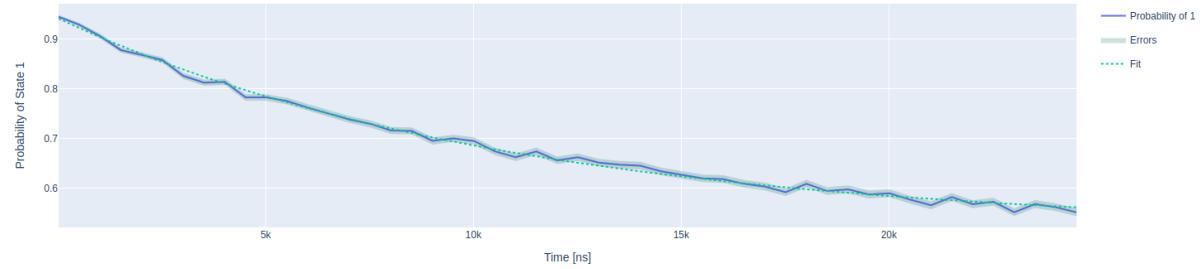
In the T_2 experiment the same pulse sequence is used with the caveat that the drive pulse is not detuned, the protocol assumes that any error on the drive frequency has already been corrected through a Ramsey experiment. For this reason the curve that describes the excited state population is expected

Figure 2.15: Output of the T_1 measurement on qubit B2.

to follow a simple exponential decay:

$$p_e(t) = A + Be^{-\frac{t}{T_2}}, \quad (2.12)$$

where A and B are two fitting parameters.

Figure 2.16: Output of the T_2 measurement on qubit B2.

Single shot classification

At this stage of the calibration procedure, the drive and measurement pulses have been individually calibrated, while a measurement now yields a clear analog output, this alone does not indicate whether the qubit was in state $|0\rangle$ or $|1\rangle$. To establish this correspondence and enable digital state assignment, it is necessary to perform a single-shot readout experiment.

To characterize the readout configuration of a superconducting qubit, the measurement is performed using a heterodyne detection setup (see [30], [15]) which acquires the transmitted microwave signal from the readout resonator. Although the average signal clearly distinguishes the two states, in a single instance, often called single shot, the signal is noisy. This noise arises both from quantum fluctuations intrinsic to the electromagnetic field inside the resonator and from added noise contributed by amplifiers and other electronics in the detection chain. The raw signal, sampled over a finite window, is digitally integrated to produce a single complex number corresponding to one measurement instance. Repeating this process after preparing the qubit in a known state produces a cloud of points in the complex IQ plane [15]. For both state $|0\rangle$ and state $|1\rangle$ we typically observe one Gaussian-shaped distribution of IQ values. The separation between these two clusters originates from the dispersive interaction between the qubit and the resonator, which induces a state-dependent frequency shift and consequently a distinguishable phase shift in the transmitted microwave tone. Since the acquisition integrates the signal over time, each shot produces a single point whose location reflects the qubit state but is blurred by noise that generates the Gaussian distributions.

The readout calibration experiment starts by preparing the qubit in state $|0\rangle$, letting it relax to its ground state and collecting IQ data without averaging. The experiment is then repeated after applying a π -pulse to prepare the qubit in state $|1\rangle$ and the unaveraged IQ-values are recorded. Plotting all these single shots in the IQ plane reveals two roughly circular clusters.

To classify the measured IQ points, the centroids of the two Gaussian-like distributions corresponding to the qubit prepared in $|0\rangle$ and $|1\rangle$ are first identified. The axis passing through these centroids is used to define the direction along which the state information is most distinguishable. The IQ plane is then rotated so that this axis aligns with the horizontal (real) axis, concentrating the relevant state-dependent variation along a single dimension. This rotation is characterized by an angle, measured in radians, between the centroid-connecting axis and the original Q -axis.

Following the rotation, the coordinate system is translated such that the centroid of the ground state distribution is placed at the origin. All IQ points are then projected onto the real axis, and the resulting one-dimensional distributions for each state are analyzed. The cumulative distribution functions of these projections are computed, and the optimal threshold is determined by finding the point that maximizes the absolute difference between the two distributions. This threshold defines the decision boundary used to assign a binary qubit state to any new measurement, based on its projection along the rotated axis.

The quality of the classification is then quantified by the assignment fidelity, defined as [30]

$$\mathcal{F} = -\frac{1}{2}(P(0,1) + P(1,0)), \quad (2.13)$$

where $P(i,j)$ is the probability of measuring the qubit in state i but prepared in state j ; this fidelity is typically expected to exceed 90–95% for a well-calibrated readout. It serves as a crucial figure of merit for the qubit, reflecting how reliably one can distinguish its quantum states in a single-shot measurement.

The result of the classification routine should be similar to the one showed in Figure (2.17).

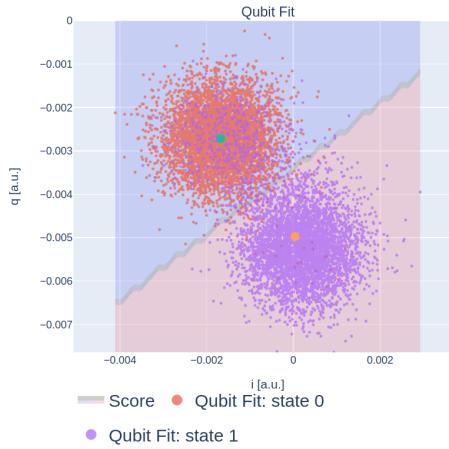


Figure 2.17: Output of the single shot classification on qubit B2.

2.2.6 Standard Randomized Benchmarking

The standard randomized benchmarking experiment will be described in more detail in the following chapter (see Section (3.1)). For now it is enough to say that in this routine, sequences of randomly selected Clifford gates are applied to the qubit, followed by an inverting gate that ideally returns the system to its initial state. By measuring the survival probability as a function of sequence length and fitting the decay curve, an estimate of the average gate fidelity is obtained. An example of the output expected from the randomized benchmarking protocol is shown in Figure (2.18).

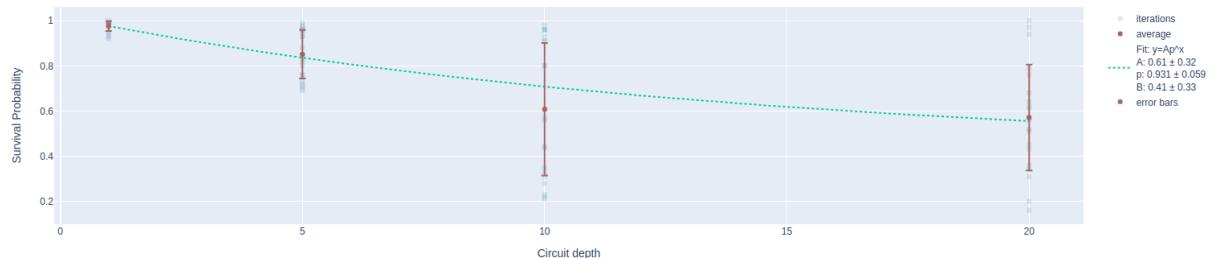


Figure 2.18: Output of the standard randomized benchmarking routine on qubit B2.

2.2.7 DRAG experiment

DRAG pulse shape

As mentioned in section (1.4), the study of optimal pulse shapes (especially for drive and control pulses) is an active area of research. The Derivative Reduction by Adiabatic Gate (DRAG) technique [27][31]

is a widely used method to mitigate leakage errors during high-fidelity single-qubit gate operations. The DRAG is the analytical solution to force the interaction Hamiltonian of the system (1.58) to be restricted to the computational space.

The DRAG scheme implements a single-qubit rotation by applying a shaped pulse with an envelope $\Omega(t)$ on one quadrature (typically along $\hat{\sigma}_x$) and a secondary pulse with an envelope proportional to the time derivative $\dot{\Omega}(t)$ on the orthogonal quadrature (typically $\hat{\sigma}_y$). For instance, a rotation about the x -axis is generated by the time-dependent Hamiltonian[7]:

$$\hat{H}_d(t) = \hbar \Omega_x(t) \sin(\omega_d t) \hat{\sigma}_x + \hbar \Omega_y(t) \sin(\omega_d t + 2\pi) \hat{\sigma}_y, \quad (2.14)$$

where the pulse shapes are defined as:

$$\Omega_x(t) = \Omega_0 e^{-t^2/(2\sigma^2)}, \quad (2.15)$$

$$\Omega_y(t) = \lambda_\eta \frac{d}{dt} \Omega_x(t), \quad (2.16)$$

where η represents the qubit anharmonicity and λ is a dimensionless scaling parameter. The efficacy of the DRAG correction arises from its ability to suppress virtual transitions that occur due to the spectral overlap of the primary pulse with off-resonant transitions in the multilevel qubit structure.

Theory predicts that setting $\lambda = 1$ minimizes the leakage to higher excited states, while $\lambda = 0.5$ is optimal for correcting phase distortions induced during the gates [27], [31], [32]. In experimental implementations, the optimal value of λ may deviate from theoretical predictions due to pulse distortions, limited bandwidth of the control electronics, and interactions with the readout resonator [33].

DRAG protocol

A straightforward experiment implemented in `Qibocal` to calibrate the λ parameter involves performing two separate measurements using DRAG pulses in the pulse sequence. In the first experiment, a sequence of DRAG pulses is applied in the order $Y_\pi X_{\frac{\pi}{2}}$ with both pulses parametrized by a given value of λ . For the second measurement the procedure is the same but the pulse order is reversed $X_\pi Y_{\frac{\pi}{2}}$. These specific sequences are chosen because they ideally result in the same final quantum state—any discrepancy between the two indicates phase misalignment, in particular these sequences exhibit opposite signs of phase errors, as explained in [34].

The post-processing consists in measuring the probability of the qubit being in the state $|1\rangle$ for each value of λ . A linear fit is performed for both pulse sequences, and the correct λ is the value where the two lines cross. This intersection corresponds to the λ value that minimizes phase errors and ensures the qubit operates with minimal distortion.

An example of the expected output for the DRAG routine is shown in Figure (2.19)

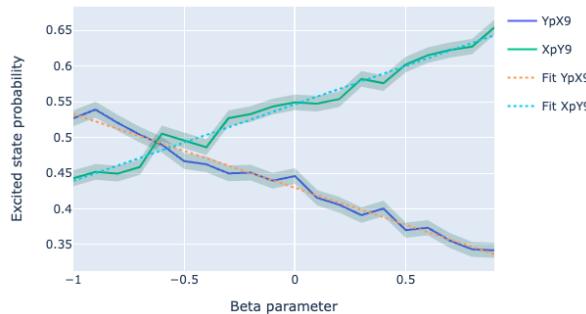


Figure 2.19: Expected output of the DRAG routine. β is the symbol used in `Qibocal` for the λ parameter.

2.3 Calibration results

In Table (2.1), are summarized the results obtained for the calibration of line D of the `qw11q` chip. The readout and assignment fidelities were obtained by performing a classification experiment on each qubit with 5000 shots, while $T1$ and $T2$ measurements were carried out using the respective protocols, each with 2048 shots.

The standard randomized benchmarking was performed at four different circuit depths: $n = 1, 5, 10, 20$; for each depth the circuit was repeated 20 times with 100 shots per repetition. The number of shots in the randomized benchmarking experiment was deliberately limited, as this routine is inherently very time-consuming.

Qubit	Readout Fidelity	Assignment Fidelity	T1 [μs]	T2 [μs]	Gate infidelity ($\cdot 10^{-3}$)
D1	0.876	0.938	26.4 ± 0.4	13.0 ± 2.0	27.0 ± 21.0
D2	0.945	0.973	14.9 ± 0.1	18.0 ± 10.0	20.5 ± 6.2
D3	0.905	0.952	23.6 ± 0.3	28.0 ± 12.0	7.0 ± 18
D4	0.929	0.964	20.6 ± 0.2	40.0 ± 5.2	4.4 ± 4.8
B2	0.929	0.964	20.6 ± 0.2	40.0 ± 5.2	4.4 ± 4.8

Table 2.1: Single qubit gates calibration results. D line was the fist one I tried to calibrate. For completeness in the table I report also the values for B2 which was calibrated lately for the purpose of showing calibration results in this chapter.

The values of the assignment fidelity and readout fidelity are overall satisfactory, indicating reliable state discrimination during measurement. Similarly, the relaxation time T_1 of the qubit shows acceptable values.

On the other hand, the coherence time T_2 is generally low, especially when considering that its theoretical upper bound is $2T_1$. This indicates the presence of significant pure dephasing mechanisms, which are limiting the coherence independently of energy relaxation processes. A reduced T_2 can have important consequences for subsequent experiments, particularly impacting the performance of randomized benchmarking protocols. Specifically, a short coherence time may degrade the fidelity of gate operations, leading to an overestimation of the average error per gate.

Undoubtedly, individuals with greater experience in the calibration of superconducting qubits are able to achieve significantly better figures of merit than those reported in table (2.1). For comparison, the current state-of-the-art in superconducting qubit calibration typically yields $T_1 = 50 - 300\mu\text{s}$ [35] and $T_2 = 50 - 300\mu\text{s}$ [35], assignment fidelity of and readout fidelities between 95% and 99.5% [15] and single gate fidelities in the range 99.5% – 99.99%.

The satisfactory, yet overall limited, results obtained from this initial calibration can be partially attributed to a non-optimal calibration procedure and partially to limitations of the experimental setup, which could be further investigated.

Chapter 3

RB fidelity optimization

Accurate control of single-qubit gates is a fundamental requirement for quantum computing. The single-gate fidelity quantifies the average performance of individual gates and directly impacts the error rates of quantum circuits and is a critical figure of merit.

The calibration procedure detailed in Section (2.2) typically requires significant time and experimental expertise, especially to obtain state-of-the art calibration. In my work, the calibration started from a pre-existing, though non optimal, configuration and the task consisted of improving the quality the gates by using available *Qibocal* protocols. Even under these conditions, the effort required to refine gate performance highlighted the limitations of manual methods.

As quantum processors scale in size and complexity, manual calibration becomes increasingly impractical. Frequent recalibrations are necessary to mitigate the effects of parameter drift and environmental fluctuations, especially in superconducting devices [15].

In this chapter, I present the results of an optimization procedure aimed at enhancing the fidelity of single-qubit gates on a superconducting quantum device. The focus was placed on the optimization of single-qubit gate fidelity, as it represents both a fundamental and relatively accessible figure of merit. High single-qubit gate fidelities are essential for the reliable construction of quantum circuits, and their characterization and improvement are comparatively simpler than for two-qubit gates. Moreover, considering the final goal of developing automated calibration procedures, a natural first step is to automate the re-calibration and optimization of existing gate parameters, building on a pre-existing calibration. Starting from single-qubit gates thus provides a practical and effective testbed for validating optimization algorithms before extending them to more complex multi-qubit operations.

3.1 Randomized Benchmarking

A strong limitation to the realization of quantum computing technologies is the loss of coherence that happens as a consequence of the application of many sequential quantum gates to the qubits. A possible approach to characterize gate error is the quantum process tomography which allows the experimenter to establish the behaviour of a quantum gates; the main drawback of this approach is that process tomography can be very time consumig since its time complexity scales exponentially with the number of qubits involved [36] and the result is affected by state preparation and measurements (SPAM) errors.

To overcome these limitations, randomized benchmarking (RB) was introduced and is currently widely used to quantify the avarage error rate for a set of quantum gates.

The main idea is that the error obtained from the combined action of random unitary gates drawn from a uniform distribution with respect to the Haar measure [37] and applied in sequence to the qubit will avarage out to behave like a depolarizing channel [38]. This last consideration simplifies the characterization of noise because it removes dependence on specific error structures and allows fidelity to be extracted through a simple exponential decay.

It was later shown that it is possible to simplify this procedure even more, by restricting the unitaries to gates in the Clifford group ¹ and by not requiring that the sequence is strictly self-inverting [39]. The fundamental principle of RB is the application of sequences of randomly selected quantum gates from the Clifford group \mathcal{C} followed by an inversion gate which, in absence of noise, return the system to its initial state. For real systems, where noise is present, the observed survival probability provides an estimate of the avarage gate fidelity.

¹unitary rotations mapping the group of Pauli operators in itself

The standard RB protocols consist of the following steps:

1. Initialize the system in ground state $|0\rangle$
2. For each sequence-length m build a sequence of m randomly drawn Clifford gates C_1, C_2, \dots, C_m
3. Determine the inverse gate $C_{m+1} = (C_m \circ \dots \circ C_1)^{-1}$
4. Measure $C_{m+1} \circ C_m \circ \dots \circ C_1 |0\rangle$

The process must be repeated for multiple sequence of the same length and with varying length.

In ideal systems without noise we should have

$$C_{m+1} \circ C_m \circ \dots \circ C_1 |0\rangle = (C_m \circ \dots \circ C_1)^{-1} \circ (C_m \circ \dots \circ C_1) |0\rangle = |0\rangle. \quad (3.1)$$

In real systems, where noise is present, eq. (3.1) does not hold; instead randomization with Clifford gates behave as a depolarizing channel (1.79) with depolarization probability d . The survival probability of the initial state $|0\rangle$ for different sequence lengths m follows the exponential decay model

$$F(m) = Ap^m + B, \quad (3.2)$$

where $1 - p$ is the rate of depolarization and A and B capture the state preparation and measurement errors.

The parameter p is directly related to the depolarization probability d through the average gate fidelity F which, for a depolarizing channel, is given by

$$F = 1 - \frac{d}{2^n - 1}. \quad (3.3)$$

For the details of the calculations to obtain eq. (3.3) see Appendix C.

Now we can derive the average error per Clifford gate $\varepsilon_{Clifford}$

$$\varepsilon_{Clifford} = 1 - F, \quad (3.4)$$

where F is the average gate fidelity. Substituting in (3.4) the formula for the average gate fidelity (3.3) we obtain

$$\varepsilon_{Clifford} = \frac{d}{2^n - 1} = \frac{1 - p}{1 - 2^{-n}}, \quad (3.5)$$

which shows how the average error per Clifford gate is directly connected to the exponential decay rate.

QUA Randomized Benchmarking

For the results we present in the following the technique used slightly differs from the one described in section (3.1)

3.2 Scipy optimization methods

3.2.1 Algorithm description

I primi metodi che abbiamo provato per l'ottimizzazione dei parametri sono quelli standard implementati nella libreria **Scipy** [40].

The first gradient-free optimization method to be tested was Nelder-Mead since in letteratura era già stato riportato il suo utilizzo per obiettivi simili [41].

The Nelder-Mead optimization method, originally introduced by Nelder and Mead in 1965 [42], is a widely used numerical optimization technique for unconstrained problems in multidimensional spaces. This derivative-free method operates using simplex, which is a polytope of $n + 1$ vertices in a n -dimensional space. The algorithm iteratively updates the simplex by replacing its worst-performing vertex with a new candidate point, thereby guiding the search towards an optimal solution. If the goal is to minimize a given function $f(\mathbf{x})$ where $\mathbf{x} \in \mathbb{R}^n$ the algorithms proceeds with the following steps:

1. If not otherwise initialized, $n + 1$ points are sampled for building the initial simplex
2. Order the test points according to their values at vertices: $f(\mathbf{x}_1) \leq f(\mathbf{x}_2) \leq \dots \leq f(\mathbf{x}_{n+1})$ and check whether the algorithm should terminate.

3. Calculate \mathbf{x}_0 , the centroid of all points except \mathbf{x}_{n+1} .
4. **Reflection:** Compute the reflected point $\mathbf{x}_r = \mathbf{x}_0 + \alpha(\mathbf{x}_0 - \mathbf{x}_{n+1})$ with $\alpha > 0$. If \mathbf{x}_r satisfies $f(\mathbf{x}_1) \leq f(\mathbf{x}_r) < f(\mathbf{x}_n)$, then a new simplex is obtained by replacing the worst-performing point \mathbf{x}_{n+1} with \mathbf{x}_r and then go to step 1.
5. **Expansion:** If \mathbf{x}_r is the current best point, meaning that $f(\mathbf{x}_r) < f(\mathbf{x}_1)$, then the expanded point is computed: $\mathbf{x}_e = \mathbf{x}_0 + \gamma(\mathbf{x}_r - \mathbf{x}_0)$ with $\gamma > 1$. If \mathbf{x}_e satisfies $f(\mathbf{x}_e) < f(\mathbf{x}_r)$, then a new simplex is obtained by replacing \mathbf{x}_{n+1} with the expanded point \mathbf{x}_e and then go to step 1. If instead $f(\mathbf{x}_e) \geq f(\mathbf{x}_r)$, the new simplex is obtained by replacing \mathbf{x}_{n+1} with \mathbf{x}_r , and then go to step 1.
6. **Contraction:** In this case is certain that $f(\mathbf{x}_r) \geq f(\mathbf{x}_n)$ then:
 - If $f(\mathbf{x}_r) < f(\mathbf{x}_{n+1})$: compute the contracted point $\mathbf{x}_c = \mathbf{x}_0 + \rho(\mathbf{x}_r - \mathbf{x}_0)$ with $0 < \rho \leq 0.5$. If \mathbf{x}_c satisfies $f(\mathbf{x}_c) < f(\mathbf{x}_r)$, then a new simplex is obtained by replacing \mathbf{x}_{n+1} with \mathbf{x}_c and go to step 1. Else go to step 6.
 - If $f(\mathbf{x}_r) \geq f(\mathbf{x}_{n+1})$: compute the contracted point $\mathbf{x}_c = \mathbf{x}_0 + \rho(\mathbf{x}_{n+1} - \mathbf{x}_0)$ with $0 < \rho \leq 0.5$. If \mathbf{x}_c satisfies $f(\mathbf{x}_c) < f(\mathbf{x}_{n+1})$, the a new simplex is constructed with \mathbf{x}_c and go to step 1. Else go to step 6.
7. **Shrinkage:** Replace all points except the best, \mathbf{x}_1 , with $\mathbf{x}_i = \sigma(\mathbf{x}_i - \mathbf{x}_1), 0 < \sigma \leq 0.5$

The algorithm terminates when the standard deviation of the function values of the current simplex fall below a user-initialized tolerance. When the cycle stops the point of the simplex associated to the lower function value is returned as proposed optimum

The values of the parameters α, γ, ρ and σ were left to default of `scipy`: $\alpha = 1, \gamma = 2, \rho = 0.5, \sigma = 0.5$.

Per valutare eventuali miglioramenti nella performance abbiamo provato ad utilizzare un algoritmo che fosse gradient-based. Nello specifico ho provato ad utilizzare l'algoritmo di Sequential Least Squares Programming (SLSQP) nella versione implementata all'interno della libreria `scipy`.

3.2.2 Results

In the following I present the results obtained provando ad effettuare l'ottimizzazione con l'algoritmo di Nelder-Meads

3.3 CMA-ES

3.3.1 Algorithm description

Covariance Matrix Adaptation Evolution Strategy (CMA-ES [43]), is a population-based evolutionary algorithm designed for optimizing complex, non-convex, and high-dimensional functions.

It belongs to the broader class of Evolution Strategies (ES), a subset of Evolutionary Algorithms (EAs)(see [44]), and is particularly effective for black-box optimization where gradient information is unavailable.

Evolution Strategies (ES) are a class of optimization methods that employ self-adaptive mechanisms to explore the search space efficiently. Unlike classical optimization techniques that rely on gradient descent, ES leverage stochastic sampling to navigate rugged and multimodal landscapes. In this context, CMA-ES is an adaptive stochastic search method that iteratively refines a probability distribution over the search space. Unlike traditional Genetic Algorithms (GAs), which rely on crossover and mutation operators, CMA-ES employs a multivariate normal distribution to generate candidate solutions. The method adaptively updates the distribution's mean and covariance matrix based on the fitness of sampled points.

The fundamental idea behind CMA-ES is the use of a multivariate Gaussian distribution to model promising search directions. Let μ_t denote the mean of the distribution at iteration t , and Σ_t the covariance matrix. Then, a new population of λ candidate solutions $\mathbf{x}_i^{(t+1)} \sim \mu_t + \sigma_t \mathcal{N}(0, \Sigma_t)$, where σ_t is a step size controlling the exploration.

The CMA-ES algorithm follows the following steps:

1. If not otherwise specified, the initial parameters are set: mean vector μ_0 , covariance matrix Σ_0^2 , step size σ_0 , population size λ
2. Generate λ new candidate solutions \mathbf{x}_i according to a multivariate normal distribution.
3. Evaluate the objective function $f(\mathbf{x}_i)$ for each candidate solution.
4. Sort the new candidate solutions based on fitness: $f(\mathbf{x}_0) \leq \dots \leq f(\mathbf{x}_\lambda)$.
5. Update the mean vector μ with the $m = \lfloor \lambda/2 \rfloor$ top performing solutions:

$$\mu \leftarrow \sum_{i=0}^m \mathbf{w}_i \mathbf{x}_i, \quad (3.6)$$

where \mathbf{w}_i are internally defined weights.

6. Update the isotropic and anisotropic evolution path $\mathbf{p}_\sigma, \mathbf{p}_c$ ³.
7. Update the covariance matrix:

$$C \leftarrow (1 - c_1 - c_\mu)C + c_1 \mathbf{p}_c \mathbf{p}_c^T + c_\mu \sum_{i=1}^\mu w_i \mathbf{y}_i \mathbf{y}_i^T, \quad (3.7)$$

where c_1 and c_μ are learning rates and \mathbf{y}_i represents the deviation of the i -th candidate solution from the mean \mathbf{mu} .

8. Update the step size using a cumulative path evolution mechanism

$$\sigma \leftarrow \sigma \cdot \exp \left(\frac{c_\sigma}{d_\sigma} (\|\mathbf{p}_\sigma\| - E\|\mathcal{N}(0, I)\|) \right), \quad (3.8)$$

where c_σ is the learning rate for step-size adaptation, d_σ is a damping factor $\|\mathbf{p}_\sigma\|$ is the length of the evolution path and $E\|\mathcal{N}(0, I)\|$ is the expected length of a standard normally distributed random vector.

Nel seguito, a meno che non sia diversamente specificato, i parametri sono stati inizializzati ai valori di default della libreria CMA-ES

3.3.2 Results

3.4 Optuna

3.4.1 Algorithm description

In addition to the optimization methods mentioned earlier, the Tree-Structured Parzen Estimator (TPE) method was employed, using its implementation available in the `optuna` library [45].

Tree-Structured Parzen Estimator (TPE) is a Sequential Model-Based Optimization (SMBO) approach [46]. SMBO methods sequentially construct models to approximate the performance of optimization parameters based on historical measurements, and then subsequently choose new parameters values to test based on this model. [47] At the heart of SMBO is the idea of building a surrogate model, which is used to predict the objective function's values for unseen parameters configurations. The surrogate model is iteratively updated as new observations are made, and the optimization process balances exploration, which focuses on uncertain regions of the search space, and exploitation, which focuses on areas that are more likely to improve the objective based on past evaluations. This balance ensures that the optimization process makes efficient use of resources and avoids wasting time on suboptimal regions.

The TPE algorithm is a probabilistic model-based optimization method that uses non-parametric density estimation to guide the search. The TPE algorithm differs from traditional Bayesian optimization approaches, such as Gaussian Process-based methods, in its modeling strategy. Rather than directly approximating the objective function, TPE constructs two separate probabilistic models:

² $\Sigma_0 = \mathbb{I}$ for isotropic search

³For details on the update process of the evolution paths see [43].

- $p(x|y < y^*)$, the likelihood of observing a parameter configuration x given that the objective function value y is below a chosen threshold y^* .
- $p(x|y \geq y^*)$, the likelihood of observing X for less promising function values.

These probability densities are estimated using non-parametric methods such as kernel density estimation (KDE). New candidate points are then generated by sampling from $p(x|y < y^*)$, favoring configurations that are expected to yield lower objective values. The threshold y^* is typically set as a quantile of observed values, ensuring a focus on the most promising regions of the search space.

The TPE method is the default optimization strategy in `Optuna`, an advantage in the optimization algorithm as implemented in `optuna` is the addition of an automatic *pruning* mechanism that stops unpromising trials early, which can significantly speed up the optimization process by avoiding unnecessary computations. In our case, this is particularly relevant because the execution of the RB routine, which is performed at each call to the cost function, requires [insert approximate execution time]

As implemented in our code, the default pruner used is the median pruner `optuna.pruners.MedianPruner`. This pruner works by evaluating the intermediate results of a trial and comparing them to the median of completed trials at the same step. If the current trial's performance is worse than the median, it is pruned to prevent wasting computational resources on unpromising configurations.

3.4.2 Results

D1 - steps

D1 - 1000 steps

D2 - 1000 steps

3.5 RB optimization conclusions

Chapter 4

Pulses analysis and tuning

Having concluded that closed-loop optimization would not significantly improve fidelity, we shifted our focus towards improvement and implementation of individual protocols to improve the accuracy of qubit operations.

In this chapter, I present the results of two additions to the `Qibocal` software. The first is the inclusion of an *RX90* gate as a native gate, which can enhance the performance of protocols requiring qubit rotations of $\frac{\pi}{2}$. The second is the implementation of the cryoscope, a routine first described in [13], which is useful for correcting distortions in the magnetic flux pulse applied to the SQUID.

4.1 RX90 calibration

4.2 Flux pulse correction

4.2.1 Cryoscope

The experiment that we describe in this section was first introduced in [13], the goal is to determine predistortions that needs to be applied to a flux pulse signal so that the qubit receives the flux pulse as intended by the experimenter.

As explained in section (1.3), accurate dynamical control of qubit frequency is of key importance to realize single- and two-qubit gates. One of the on-chip control variable that is used on QunatumWare chip is the magneteic flux through a SQUID loop, the signal for magnetic flux control originates from an arbitrary waveform generator (AWG) which operates at room temperature.

As the signal propagates through various electrical components along the control line leading to the quantum device it undergoes linear dynamical distortions. If not properly compensated, these distortions can degrade gate performance, impacting experiments fidelity and repeatability.

In [13] is proposed a technique to characterize flux-pulse distortions induced by components inside the dilution refrigerator by directly measuring the qubit state. In this protocol we send the qubit a pulse sequence where a flux pulse of varying duration τ is embedded between two $\frac{\pi}{2}$ pulses which are always separated by a fixed interval T_{sep} .

The first $\frac{\pi}{2}$ pulse rotates the qubit of $\frac{\pi}{2}$ around the y axis of the Bloch sphere changing its state from $|0\rangle$ to $\frac{|0\rangle+|1\rangle}{\sqrt{2}}$.

When a flux pulse $\Phi_{Q,\tau}(t)$ of duration τ is sent to the qubit¹ after the first $\frac{\pi}{2}$ pulse, the qubits evolve to the state $\frac{|0\rangle+e^{i\varphi\tau}|1\rangle}{\sqrt{2}}$ with relative quantum phase

$$\frac{\varphi_\tau}{2\pi} = \int_0^{T_{sep}} \Delta f_Q(\Phi_{Q,\tau(t)}) dt = \int_0^\tau \Delta f_Q(\Phi_{Q,\tau(t)}) dt + \int_\tau^{T_{sep}} \Delta f_Q(\Phi_{Q,\tau(t)}) dt \quad (4.1)$$

where in the second step we separated the contributions from flux response up to τ and the turn-off transient.

The experiment is then completed with a $\frac{\pi}{2}$ rotation aroud the y - or x -axis of the Bloch sphere to measure respectively the $\langle X \rangle$ or $\langle Y \rangle$ components of the Bloch vector when applying the measurement

¹To send a $\Phi_{Q,\tau}(t)$ flux pulse we are actually sending a $V_{in,\tau}(t)$ voltage pulse through the electronics

gate MZ . From the measurement of $\langle X \rangle$ and $\langle Y \rangle$ we can extract the relative phase ϕ_τ .

Then we can estimate $\Phi_Q(t)$ in the interval $[\tau, \tau + \Delta\tau]$ as follows. From the measurement of $\phi_{\tau+\Delta\tau}$ and ϕ_τ we can compute $\overline{\Delta f_R}$:

$$\overline{\Delta f_R} = \frac{\phi_{\tau+\Delta\tau} - \phi_\tau}{2\pi\Delta\tau} \quad (4.2)$$

$$= \frac{1}{\Delta\tau} \left(\int_0^{\tau+\Delta\tau} \Delta f_Q(\Phi_{Q,\tau+\Delta\tau}(t)) dt + \int_{\tau+\Delta\tau}^{T_{sep}} \Delta f_Q(\Phi_{Q,\tau+\Delta\tau}(t)) dt \right) \quad (4.3)$$

$$- \frac{1}{\Delta\tau} \left(\int_0^\tau \Delta f_Q(\Phi_{Q,\tau}(t)) dt - \int_\tau^{T_{sep}} \Delta f_Q(\Phi_{Q,\tau}(t)) dt \right) \quad (4.4)$$

$$= \frac{1}{\Delta\tau} \left(\int_\tau^{\tau+\Delta\tau} \Delta f_Q(\Phi_{Q,\tau+\Delta\tau}(t)) dt + \int_{\tau+\Delta\tau}^{T_{sep}} \Delta f_Q(\Phi_{Q,\tau+\Delta\tau}(t)) dt - \int_\tau^{T_{sep}} \Delta f_Q(\Phi_{Q,\tau}(t)) dt \right) \quad (4.5)$$

$$= \frac{1}{\Delta\tau} \int_\tau^{\tau+\Delta\tau} \Delta f_Q(\Phi_{Q,\tau+\Delta\tau}(t)) dt + \varepsilon \quad (4.6)$$

with

$$\varepsilon = \frac{1}{\Delta\tau} \left(\int_{\tau+\Delta\tau}^{T_{sep}} \Delta f_Q(\Phi_{Q,\tau+\Delta\tau}(t)) dt - \int_\tau^{T_{sep}} \Delta f_Q(\Phi_{Q,\tau}(t)) dt \right)$$

The phase contribution from the turn-off transients is minimal due to the sharp return to the first-order flux-insensitive sweet spot of the nearly quadratic $\Delta f_Q(\Phi_Q)$; numerical simulations suggest that $|\varepsilon|/\Delta f_R$ remains within the range of approximately 10^{-2} to 10^{-3} for typical dynamical distortions in commonly used electronic components[48][49], for this reason it will be neglected.

Then we can obtain the reconstructed flux pulse $\Phi_R(t)$ inverting eq. (1.32).

Pulse reconstruction

Corrections study

4.2.2 Corrected pulse

4.2.3 Filter determination

IIR corrections

FIR corrections

Una volta trovati i feedback e feedforward taps per gli IIR filters, è necessario trovare gli FIR per correggere il segnale su scale di tempo più brevi.

for description and notes on CMA-ES see section (3.2)

Output filters in QM

Each analog output port of the OPX system used in this work is equipped with a digital filter that is applied to the signal in the digital domain before conversion to analog. The filters are provided to the OPX via the `parameters.json` file, which contains the coefficients for the feedforward and feedback components according to the equation

$$y[n] = \sum_{m=1}^M a_m y[n-m] + \sum_{k=0}^K b_k x[n-k], \quad (4.7)$$

where $y[n]$ is the output signal, $x[n]$ is the input waveform, a_m are the feedback coefficients, and b_k are the feedforward coefficients.

In our case, we have a set of feedback coefficients determined through IIR correction and two sets of feedforward coefficients: the first obtained from the IIR-based correction, and the second from FIR-based correction on short timescales. To uniquely determine the coefficients to be passed to the electronics, it is necessary to derive a single set of feedforward coefficients and a single set of feedback coefficients by combining the two sets of feedforward filters through convolution.

Infatti posso considerare il segnale di input x a cui applico un primo filtro IIR ottenendo così un segnale in output y tale che

$$y[n] = \sum_{m=1}^M a[m]y[n-m] + \sum_{k=0}^N b[k]x[n-k]. \quad (4.8)$$

al segnale y applico un secondo filtro IIR ottenendo quindi un segnale z in output con la sgeuente forma

$$z[n] = \sum_{m=1}^M a'[m]z[n-m] + \sum_{k=0}^N b'[k]y[n-k]. \quad (4.9)$$

Now I consider equation (4.8) and rewrite it as

$$y[n] - \sum_{m=1}^M a[m]y[n-m] = \sum_{k=0}^N b[k]x[n-k], \quad (4.10)$$

then by applying a Z-transform we obtain

$$Y(z) \left(1 - \sum_{m=1}^M a[m]z^{-m} \right) = X(z) \left(\sum_{k=0}^N b[k]z^{-k} \right) \quad (4.11)$$

so that

$$H_1(z) = \frac{Y(z)}{X(z)} = \frac{\sum_{k=0}^N b[k]z^{-k}}{1 - \sum_{m=1}^M a[m]z^{-m}} = \frac{B(z)}{A(z)} \quad (4.12)$$

$$\rightarrow Y(z) = H_1(z)X(z) = \frac{B(z)}{A(z)}X(z) \quad (4.13)$$

$$(4.14)$$

We can do the same also for equation (4.9) and rewrite it as

$$z[n] = \sum_{m=1}^M a'[m]z[n-m] + \sum_{k=0}^N b'[k]y[n-k], \quad (4.15)$$

which, by applying the Z-transform becomes

$$Z(z) \left(1 - \sum_{m=1}^M a'[m]z^{-m} \right) = Y(z) \left(\sum_{k=0}^N b'[k]z^{-k} \right). \quad (4.16)$$

Again we can write the the transfer function

$$H_2(z) = \frac{Z(z)}{Y(z)} = \frac{\sum_{k=0}^N b'[k]z^{-k}}{1 - \sum_{m=1}^M a'[m]z^{-m}} = \frac{B'(z)}{A'(z)} \quad (4.17)$$

$$\rightarrow Z(z) = H_2(z)Y(z) = \frac{B'(z)}{A'(z)}Y(z) = \frac{B'(z)}{A'(z)} \frac{B(z)}{A(z)}X(z) \quad (4.18)$$

$$= \left(\frac{\sum_{k=0}^N b'[k]z^{-k}}{1 - \sum_{m=1}^M a'[m]z^{-m}} \right) \left(\frac{\sum_{k=0}^N b[k]z^{-k}}{1 - \sum_{m=1}^M a[m]z^{-m}} \right) X(z) \quad (4.19)$$

$$(4.20)$$

From the transfer function we can obtain the expression for $Z(z)$ in terms of $X(z)$

$$\rightarrow Z(z) \left(1 - \sum_{m=1}^M a'[m]z^{-m} \right) \left(1 - \sum_{m=1}^M a[m]z^{-m} \right) = \left(\sum_{k=0}^N b'[k]z^{-k} \right) \left(\sum_{k=0}^N b[k]z^{-k} \right) X(z) \quad (4.21)$$

$$\rightarrow Z(z) \left(\sum_{m=0}^M a'[m]z^{-m} \right) \left(\sum_{m=0}^M a[m]z^{-m} \right) = \left(\sum_{k=0}^N b'[k]z^{-k} \right) \left(\sum_{k=0}^N b[k]z^{-k} \right) X(z) \quad (4.22)$$

$$(4.23)$$

where in the last step ho assorbito l'1 in the sum come coeffiente a_0 . By expanding the products we obtain

$$\left(\sum_{m=0}^M a'[m] \right) \left(\sum_{m=0}^M a[m] \right) = \sum_{m=0}^{2M} \sum_{i=0}^m a'[i]a[m-i] = c[k], \quad \text{with } m = 0, \dots, 2M, \quad (4.24)$$

$$\left(\sum_{k=0}^N b'[k] \right) \left(\sum_{k=0}^N b[k] \right) = \sum_{k=0}^{2N} \sum_{i=0}^k b[i]b[k-i] = d[k], \quad \text{with } k = 0, \dots, 2N. \quad (4.25)$$

$$(4.26)$$

It is then possible to re-write equation (4.23) using the new expression for the filters

$$Z(z) \left(\sum_{m=0}^{2M} c[m]z^{-m} \right) = \left(\sum_{k=0}^{2N} d[k]z^{-k} \right) X(z) \quad (4.27)$$

$$\rightarrow Z(z) \left(1 - \sum_{m=1}^{2M} c[m]z^{-m} \right) = \left(\sum_{k=0}^{2N} d[k]z^{-k} \right) X(z) \quad (4.28)$$

$$(4.29)$$

then we apply the inverse-Z-transform and obtain

$$z[n] - \sum_{m=1}^{2M} c[m]z[n-m] = \sum_{k=0}^{2N} d[k]x[n-k] \quad (4.30)$$

$$z[n] = \sum_{m=1}^{2M} c[m]z[n-m] + \sum_{k=0}^{2N} d[k]x[n-k] \quad (4.31)$$

$$(4.32)$$

where the feedforward (feedback) taps of the final filters, are given by the convolution of the feedforward (feedback) taps of the two filters as shown in equations (4.25) and (4.26).

4.2.4 Dimostrazione del conto

In general, for different forms of the detuning flux $\Delta f(\Phi) = a\Phi^k$, where $k \in \mathbb{Z}^+$, the phase φ_τ expressed in terms of the impulse response $h = \frac{ds}{dt}$ is the following,

$$\varphi_\tau = 2\pi a \int_0^\infty \left[\int_0^\infty h(t-t')dt' - \int_0^\infty h(t-\tau-t')dt' \right]^k dt \quad (4.33)$$

$$= 2\pi a \int_0^\tau \left[\int_0^t h(t-t')dt' \right]^k dt + 2\pi a \int_\tau^\infty \left[\int_0^\tau h(t-t')dt' \right]^k dt, \quad (4.34)$$

The demonstration of this equality is reported in Appendix B.

Chapter 5

Conclusions

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Appendix A

Appendix B

In the following we will demonstrate the correctness of equation (4.33):

$$\varphi_\tau = 2\pi a \int_0^\infty \left[\int_0^\infty h(t-t')dt' - \int_0^\infty h(t-\tau-t')dt' \right]^k dt \quad (5.1)$$

$$= 2\pi a \int_0^\tau \left[\int_0^t h(t-t')dt' \right]^k dt + 2\pi a \int_\tau^\infty \left[\int_0^\tau h(t-t')dt' \right]^k dt, \quad (5.2)$$

for different forms of the detuning flux

$$\Delta f(\Phi) = a\Phi^k \quad (5.3)$$

where $k \in \mathbb{Z}^+$. From the calculations showed in (4.1) we know that the relative phase ϕ_τ for a general form of the detuning flux (5.3), with $T_{\text{sep}} = \infty$ is

$$\varphi_\tau = 2\pi \int_0^{+\infty} a [(s(t) - s(t-\tau))]^k dt = 2\pi a \int_0^{+\infty} [(s(t) - s(t-\tau))]^k dt \quad (5.4)$$

As hypothesis we know that voltage-to-flux step response of the control line is

$$s(t) = \left(1 - e^{-t/\tau}\right) \cdot u(t), \quad (5.5)$$

where $u(t)$ is the step function, and that the impulse response is

$$h(t) = \frac{ds}{dt} \quad (5.6)$$

If we substitute the expression of $s(t)$ given in (5.5) in equation (5.4) we obtain

$$\varphi_\tau = 2\pi a \int_0^{+\infty} \left[\int_0^{+\infty} h(t-t')dt' - \int_0^{+\infty} h(t-\tau-t')dt' \right]^k dt. \quad (5.7)$$

To do we have to show that

1.

$$s(t) = \int_0^{+\infty} h(t-t')dt' \quad (5.8)$$

2.

$$s(t-\tau) = \int_0^{+\infty} h(t-\tau-t')dt' \quad (5.9)$$

We start from the demonstration of equation (5.8). By definition (5.6) we can write

$$h(t) = \frac{d}{dt} \left[\left(1 - e^{-t/\tau}\right) u(t) \right] = \frac{e^{-t/\tau}}{\tau} u(t) + \left(1 - e^{-t/\tau}\right) \delta(t), \quad (5.10)$$

substituting Eq (5.10) in Eq (5.8) we obtain

$$\int_0^{+\infty} h(t-t')dt' = \int_0^{+\infty} \frac{e^{-\frac{(t-t')}{\tau}}}{\tau} u(t-t')dt' + \int_0^{+\infty} \left(1 - e^{-\frac{(t-t')}{\tau}}\right) \delta(t-t')dt', \quad (5.11)$$

by setting $t'' = t - t'$, $dt'' = -dt'$, we have $t'' \rightarrow -\infty$ for $t' \rightarrow +\infty$ and $t'' = t$ for $t' = 0$, the integral then becomes

$$\int_0^{+\infty} h(t - t') dt' = \int_t^{-\infty} -\frac{e^{-t''/\tau}}{\tau} u(t'') dt'' - \int_t^{-\infty} (1 - e^{-t''/\tau}) \delta(t'') dt'' \quad (5.12)$$

$$= \int_{-\infty}^t \frac{e^{-t''/\tau}}{\tau} u(t'') dt'' + \int_{-\infty}^t (1 - e^{-t''/\tau}) \delta(t'') dt'' \quad (5.13)$$

$$= \int_0^t \frac{e^{-t''/\tau}}{\tau} u(t'') dt'' + (1 - e^{-t''/\tau}) \Big|_{t''=0} \quad (5.14)$$

$$= \left[-e^{-t''/\tau} u(t'') \right]_0^t + 0 \quad (5.15)$$

$$= (1 - e^{-t/\tau}) u(t) \quad (5.16)$$

$$(5.17)$$

that concludes the demonstration of Eq (5.8). To demonstrate equation (5.9) we start again by using the definition of $s(t)$ to compute

$$h(t - t' - \tau) = \frac{d}{dt} \left[(1 - e^{-(t-t'-\tau)/\tau}) u(t - t' - \tau) \right] \quad (5.18)$$

$$= \frac{e^{-(t-t'-\tau)/\tau}}{\tau} u(t - t' - \tau) + (1 - e^{-(t-t'-\tau)/\tau}) \delta(t - t' - \tau) \quad (5.19)$$

$$(5.20)$$

We can substitute (5.20) in equation (5.9) and obtain

$$\int_0^{+\infty} h(t - t' - \tau) dt' = \int_0^{+\infty} \frac{e^{-(t-t'-\tau)/\tau}}{\tau} u(t - t' - \tau) dt' + \int_0^{+\infty} (1 - e^{-(t-t'-\tau)/\tau}) \delta(t - t' - \tau) dt' \quad (5.21)$$

by setting $t'' = t - t' - \tau$, $dt'' = -dt'$, we have $t'' \rightarrow -\infty$ for $t' \rightarrow +\infty$ and $t'' = t - \tau$ for $t' = 0$, the integral then becomes

$$\int_0^{+\infty} h(t - t' - \tau) dt' = \int_{-\infty}^{t-\tau} \frac{-e^{-t''/\tau}}{\tau} u(t'') dt'' - \int_{t-\tau}^{\infty} (1 - e^{-t''/\tau}) \delta(t'') dt'' \quad (5.22)$$

$$= \int_{-\infty}^{t-\tau} \frac{e^{-t''/\tau}}{\tau} u(t'') dt'' + \int_{-\infty}^{t-\tau} (1 - e^{-t''/\tau}) \delta(t'') dt'' \quad (5.23)$$

$$= \int_0^{t-\tau} \frac{e^{-t''/\tau}}{\tau} u(t'') dt'' + (1 - e^{-t''/\tau}) \Big|_{t''=0} \quad (5.24)$$

$$= \left[-e^{-t''/\tau} u(t'') \right]_0^{t-\tau} + 0 \quad (5.25)$$

$$= (1 - e^{-t''/\tau}) u(t'') \quad (5.26)$$

$$= s(t'') = s(t - t' - \tau) \quad (5.27)$$

With this, we demonstrated that

$$\varphi_\tau = 2\pi a \int_0^\infty \left[\int_0^\infty h(t - t') dt' - \int_0^\infty h(t - \tau - t') dt' \right]^k dt, \quad (5.28)$$

We now need to show that

$$2\pi a \int_0^\infty \left[\int_0^\infty h(t - t') dt' - \int_0^\infty h(t - \tau - t') dt' \right]^k dt \quad (5.29)$$

$$= 2\pi a \int_0^\tau \left[\int_0^t h(t - t') dt' \right]^k dt + 2\pi a \int_\tau^\infty \left[\int_0^\tau h(t - t') dt' \right]^k dt. \quad (5.30)$$

As first step we can try to rewrite the left-hand-side (LHS) in a different way:

$$2\pi a \int_0^\infty \left[\int_0^\infty h(t-t')dt' - \int_0^\infty h(t-\tau-t')dt' \right]^k dt \quad (5.31)$$

$$= 2\pi a \int_0^\infty [s(t) - s(t-\tau)]^k dt \quad (5.32)$$

$$= 2\pi \alpha \int_0^{+\infty} \left[(1 - e^{-t/\tau}) u(t) - (1 - e^{-(t-\tau)/\tau}) u(t-\tau) \right]^k dt \quad (5.33)$$

$$= 2\pi \alpha \int_0^\tau \left[(1 - e^{-t/\tau}) u(t) \right]^k dt + 2\pi \alpha \int_\tau^{+\infty} \left[(1 - e^{-t/\tau}) u(t) - (1 - e^{-(t-\tau)/\tau}) u(t-\tau) \right]^k dt \quad (5.34)$$

$$= 2\pi \alpha \int_0^\tau \left[\int_0^\infty h(t-t')dt' \right]^k dt + 2\pi \alpha \int_\tau^{+\infty} [s(t) - s(t-\tau)]^k dt \quad (5.35)$$

$$(5.36)$$

In the first step, to get equation (5.33) I simply used the equations (5.8) and (5.9) that were demonstarted before, then by substituting the definition of $s(t)$ we obtain Eq (5.34). It is possible to separate the integral in equation (5.34) by using the definition of $u(t)$ which is null for $t\tau$ and of $u(t-\tau)$ which is null also for $0 < t < \tau$, doing this we obtain Eq, (5.35) By substituting back (5.8) in the first term of equation (5.35) we obtain the first term of equation (5.29) which we want to demonstrate, in the second term of the sum instead, we can substitute back the definition of $s(t)$.

At this point we only have to show that for $t > \tau$, which is the interval we are considering in the second term,

$$2\pi \alpha \int_\tau^{+\infty} [s(t) - s(t-\tau)]^k dt = 2\pi a \int_\tau^\infty \left[\int_0^\tau h(t-t')dt' \right]^k dt. \quad (5.37)$$

To do this we can first evaluate $s(t) - s(t-\tau)$ in the interval $t > \tau$ so that $u(t) = u(t-\tau) = 1$, we have

$$s(t) - s(t-\tau) = 1 - e^{-\frac{t}{\tau}} - 1 + e^{-\frac{(t-\tau)}{\tau}} = e^{-\frac{t}{\tau}} (e^{\frac{\tau}{\tau}} - 1) = (e-1)e^{-\frac{t}{\tau}} \quad (5.38)$$

If we then compute $\int_0^\tau h(t-t')dt'$ we obtain

$$\int_0^\tau h(t-t')dt' = \int_0^\tau \left[\frac{e^{-\frac{(t-t')}{\tau}}}{\tau} u(t-t') - (1 - e^{-\frac{t-t'}{\tau}}) \delta(t-t') \right] dt' \quad (5.39)$$

$$= \int_0^\tau \frac{e^{-\frac{(t-t')}{\tau}}}{\tau} u(t-t')dt' \quad (5.40)$$

$$= \int_0^\tau \frac{e^{-\frac{(t-t')}{\tau}}}{\tau} dt' \quad (5.41)$$

$$= \int_0^\tau \frac{e^{-\frac{t}{\tau}} e^{\frac{t'}{\tau}}}{\tau} = e^{-\frac{t}{\tau}} e^{\frac{t'}{\tau}} \Big|_0^\tau = e^{\frac{-t}{\tau}} (e-1) \quad (5.42)$$

which is equal to (5.38) and then concludes the demonstration.

Appendix C

The average gate fidelity measures how well a noisy quantum gate \mathcal{E} approximates an ideal unitary gate U when averaged over all pure input state. It is defined as

$$F = \int d\psi \langle \psi | U^\dagger \mathcal{E}(\rho) U | \psi \rangle,$$

where the integral is taken over all the pure states $|\psi\rangle$ according to the Haar measure. In randomized benchmarking, we assume that errors behave like a depolarizing channel, meaning that after applying a noisy Clifford gate, the system is left in the correct state with probability $1 - d$ and in a completely mixed state with probability d . The quantum maps that represent a depolarizing channel is

$$\mathcal{E}(\rho) = (1 - d)U\rho U^\dagger + d\frac{\mathbb{I}}{2^n} \quad (5.43)$$

When we substitute it in the average gate fidelity definition we get:

$$\begin{aligned} F &= \int d\psi \langle \psi | U^\dagger \left[(1 - d)U |\psi\rangle \langle \psi| U^\dagger + d\frac{\mathbb{I}}{2^n} \right] U |\psi\rangle \\ &= \int d\psi \langle \psi | U^\dagger [(1 - d)U |\psi\rangle \langle \psi| U^\dagger] U |\psi\rangle + \int d\psi \langle \psi | U^\dagger \left[d\frac{\mathbb{I}}{2^n} \right] U |\psi\rangle \\ &= (1 - d) \int d\psi \langle \psi | |\psi\rangle \langle \psi| |\psi\rangle + \frac{d}{2^n} \int d\psi \langle \psi | U^\dagger \mathbb{I} U |\psi\rangle \\ &= (1 - d) + \frac{d}{2^n} \int d\psi \langle \psi | |\psi\rangle \\ &= (1 - d) + \frac{d}{2^n} = 1 - d + \frac{d}{2^n} = 1 - d \left(1 - \frac{1}{2^n} \right) \\ &= 1 - d \frac{2^n - 1}{2^n} \end{aligned}$$

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