学位論文

A Frequency-Division Multiplexing Readout System for Large-Format TES X-Ray Microcalorimeter Arrays towards Future Space Missions

将来の宇宙ミッションをめざした TES型X線マイクロカロリメータ大規模アレイの 周波数分割型信号多重化システム

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Abstract

In future space missions X-ray TES microcalorimeters are promising detectors achieving high energy resolution and high spatial resolution at the same time. It is however still difficult to realize a large format array on account of the absence of a well-established multiplexing readout technique, which is essential to suppress the heat input to a cryogenic stage with a limited cooling power by reducing the required number of wires. In this thesis, we developed the new FDM readout system aimed to realize the 400 TES microcalorimeter array for the DIOS mission, as well as large format arrays with more than a thousand of TES for future space missions. The developed system consists of the low-power SQUID, the digital FLL electronics, and the analog front-end to bridge the SQUID and the FLL electronics. The developed SQUID has very small heat dissipation and can be place at the same cryogenic stage as TES. We also developed a multi-input FDM SQUID chip using the low-power SQUID. With the digital FLL electronics that provides the high-density signal multiplexing, the cryogenic stage becomes significantly simple even for multiplexing hundreds of TES. The digital electronics can multiplex more than 16 signals without deteriorating the required SNR. Moreover, it properly reduces the data rate and keeps the required system data transfer rate to a practical number even for thousands of signal multiplexing. Using the developed readout system, we performed the TES readout experiment, and succeeded to multiplex four TES with the single-staged cryogenic setup for the first time. Although the energy resolution was poor due to the unstable ETF of TES, the feasibility of the large format TES array with the developed readout system has been demonstrated.

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Chapter 1

Introduction

In future space missions X-ray transition-edge sensor (TES) microcalorimeters are most promising detectors achieving high spectral resolution and high spatial resolution *at the same time*. The theoretical energy resolution of TES can be 2 eV (full-width at half maximum; FWHM) or better, which is an enhancement by more than a order of magnitude compared with ordinary X-ray CCD image sensors. Microcalorimeters with semiconductor thermistors, such as the Soft X-ray Spectrometer (SXS) aboard Astro-H, a next generation JAXA X-ray satellite, also provide the capability of high energy resolution ($\leq 7 \text{ eV}$ FWHM) yet the capability of high spatial resolution on account of a difficulty of multiplexing because of high-impedance inputs and parasitic capacitances of thermistors. On the other hand, a TES has a low input impedance, and is well suited for signal multiplexing that is indispensable for high spatial resolution, or a large format array, being placed at a cryogenic stage below 100 mK.

Frequency-division multiplexing (FDM) is a key technique among other methods, such as time-division multiplexing (TDM) and code-division multiplexing (CDM), for TES signal multiplexing in space missions, and is largely adopted to readout TES bolometers for not only space missions such as SAFARI in SPICA but also ground missions such as the multi-color TES bolometer camera in ASTE. It is, however, still transient and has not yet well-established for X-ray TES microcalorimeters, since the signal bandwidth of X-ray TES is much larger than that of TES bolometers. So far our research group and a research group at SRON have succeeded to multiplex signals from X-ray TES microcalorimeters.

In this thesis, we freshly developed an entire readout system including a low-power superconducting quantum interference device (SQUID), a low-noise preamplifier, and a high-frequency flux-locked loop (FLL) electronics, for large-format TES X-ray microcalorimeter arrays toward the Diffuse Intergalactic Oxygen Surveyor (DIOS) space mission as well as future space missions like ATHENA, the largest X-ray observatory proposed by the European Space Agency (ESA). Those missions are planned to carry large-format TES arrays (400 for DIOS, and 3840 for Athena), and the FDM is a must in successful science missions. Unlike the multi-staged cryogenic setups that other groups adopt, our cryogenic stage is the simple single-staged setup that all the cryoelectronics including TES and SQUIDs are placed at the very same stage. It reduces complexities of the cryogenic stage, making the required resource, the lead-time, the chance of error, and even the cost also reduced greatly. To achieve it, we developed the low-power SQUID optimized for the FDM readout, which is described in Chapter 3, and the high-frequency digital FLL electronics for FDM including the analog front-end optimized for our low-power SQUID and the digital electronics, which is described in Chapter 4. Using the developed system, we succeeded to multiplex and readout TES signals, which is described in Chapter 5. We finally summarize and conclude our system in Chapter 6.

Chapter 2

Review

2.1 Missing Baryon Problem and Next Generation X-ray Space Missions

2.1.1 Mission Baryon Problem

The observation of the cosmic microwave background by WMAP and Planck has revealed the Universe's precise ingredients. Under the Λ CDM model, atoms, or baryons, is only about 5% of the Universe. So far we have seen some of those baryons in the form of stars, galaxies, clusters of galaxies, and intercluster medium in various wavelengths. However, we have identified only 50% of them [e.g. 29]. The other half has not yet observed in any wavelengths. This is called "missing baryon problem."

According to the cosmological hydrodynamic simulation by Cen and Ostriker [3], it is indicated that some of missing baryons are distributed as a rarefied gas in a temperature of 10^5-10^7 K along the filaments interconnecting clusters of galaxies. It is called the warm-hot intergalactic medium (WHIM).



Fig. 2.1 The ion fraction distributions represented as column densities for a total gas column of 10^{19} cm^{-2} and metallicities of 0.1 Z_{\odot} . Adapted from Figure 5 of Bregman [2].



Fig. 2.2 The simulated X-ray spectra of an optically-thin thermal plasma with kT = 3 keV for various energy resolutions

As we see in Figure 2.1, the fraction of highly ionized oxygen is the largest component in the 10^{5} – 10^{7} K gas, which makes the emission/absorption lines of highly ionized oxygen, OVI, OVII and OVIII, dominated in the spectrum. Consequently, the WHIM has been searched in the UV and X-ray bands. To observe the absorption lines of OVI Ly α , we need a bright X-ray point source at the back. This makes the observable region very limited. On the other hand, the X-ray detectors in orbits are lack of the energy resolution and field of view to observe the oxygen emission lines from the WHIM. To solve the mystery of the missing baryon, a telescope with a better spatial resolution and a detector with a better energy resolution are demanded.

2.1.2 Next Generation X-ray Space Missions

X-ray Microcalorimeters

The X-ray microcalorimeter is a detector to determine the energy of an incident X-ray photon by precisely measuring a temperature increase of an absorber (or the detector itself) hit by the photon. Being placed at a cryogenic stage below 100 mK, it can achieve a very high energy resolution ($\Delta E < 10 \text{ eV}$). It is the only detector that has a sufficient energy resolution for the most important energy range of the iron K α . X-ray CCD detectors are simply lack of the required energy resolution, and energy-dispersive X-ray spectrometers, such as gratings, are not suited for diffuse X-ray sources. For this reason, they are promising detectors in future X-ray space missions.

The microcalorimeter is, in other words, a thermometer, and semiconductor thermistors have been used, but they are not suited for signal multiplexing due to the high impedance input and the high input capacitance, making it difficult to realize a large-format microcalorimeter array. The transition-edge sensor (TES) is another form of thermometer that has a higher sensitivity and a very small input impedance, and recently it has been intensively studied towards the realization of the large-format microcalorimeter array. In principle, TES can achieve $\Delta E < 2 \text{ eV}$ for the iron K α , which makes it possible to observe the detailed fine structures of the iron K α line (Figure 2.2). In our joint research group of ISAS/JAXA and Tokyo Metropolitan University, we have achieved the energy resolution of 2.8 eV FWHM for 5.9 keV X-rays [1].

For energy resolution we are achieving the requirement, yet a method to readout thousands of TES is still transient. The Soft X-ray Spectrometer of Astro-H is a 6×6 X-ray microcalorimeter with thermistors, and each pixel is read out individually. However, it is impossible to readout thousands of TES individually on account of the heat flow to the cryogenic stage through the wires. The TES signals therefore need to be multiplexed to



Fig. 2.3 The comparison of $S\Omega$ for DIOS and other space missions

reduce the required number of wires. The frequency-division multiplexing will be used to multiplex TES signals in orbits, but the method has not been well-established yet.

DIOS Mission

DIOS, the Diffuse Intergalactic Oxygen Surveyor, is a small class science satellite, proposed by the joint research group of Tokyo Metropolitan University, Nagoya University, ISAS/JAXA and the DIOS working group, aimed at unveiling the mystery of missing baryons by the direct observation of WHIM [e.g. 24]. It has been developed for a launch around 2020. It is aimed to directly detect the WHIM as well as to map the distribution of 10^5-10^7 K WHIM within 0 < z < 0.3 by observing the oxygen lines, OVII (561, 568 and 574 eV) and OVIII (653 eV), precisely. It is also aimed to reveal the chemical evolution in the intergalactic medium, the gas heating mechanism, and the gas dynamics from the emission/absorption line ratio of OVII and OVIII as well as the width and fine structure of the lines.

According to the simulation result by Yoshikawa et al. [36], 20–30 % of all the baryons can be detectable if a detector has a sensitivity of ~ $10^{11} \text{ erg}^{-1} \text{ cm}^{-2} \text{ ster}^{-1}$ to emission lines. It means that the required $S\Omega$, which is the detector effective area × the field of view, is ~ $100 \text{ cm}^2 \text{ deg}^2$ assuming the ~ 1 Msec exposure. Figure 2.3 shows the comparison of $S\Omega$ of DIOS with other space missions. Although it is a small class satellite, it has the large $S\Omega$ (~ $150 \text{ cm}^2 \text{ deg}^2$). Moreover, with the excellent energy resolution it outperforms Suzaku in the sensitivity to the diffuse X-ray emission by more than 40 times.

To achieve the science mission, the use of a TES microcalorimeter array as the detector is crucial. The required energy resolution is $\Delta E < 2 \,\mathrm{eV}@2 \,\mathrm{keV}$, which makes TES is the only option. DIOS is planned to carry a 256 to 400 TES array covering an area of ~ $1 \,\mathrm{cm}^2$. The corresponding field of view is ~ 50 arcmin. To cool the TES array and the cold font-end electronics to 50 mK, a 2-stage adiabatic demagnetization refrigerator, which is precooled by a 2-stage Stirling cooler and a ³He Joule-Thomson cooler, is used. They are all cryogen-free, thus the observing life in the orbit is unlimited.

ATHENA Mission

ATHENA, the Advanced Telescope for High ENergy Astrophysics, is a X-ray science satellite proposed by European Space Agency, scheduled for launch in 2028, and is another future satellite mission planning to use a TES array for its main detector. Three main target sciences of ATHENA are black holes and accretion physics, the cosmic feedback, and the large-scale structure of the Universe. The X-ray Integral Field Unit (X-IFU), one of the detectors aboard ATHENA, is a large-format X-ray TES microcalorimeter array, which is the key detector for the proposed science [e.g. 25].

The X-IFU covers the wide energy range of 0.2–12 keV with the excellent energy resolution of 2.5eV (E < 7 keV). The size of array is 3840, which will be multiplexed for 40 channels each, using the frequency-division multiplexing (FDM) [e.g. 26].



Fig. 2.4 A schematic view of an X-ray microcalorimeter

2.2 X-ray Microcalorimeters

2.2.1 Principles of X-ray Microcalorimeters

X-ray microcalorimeters are detectors to determine the energy of an incident X-ray photon by measuring the temperature rise caused by the absorption of the photon into an absorber. A simple microcalorimeter consists of three parts: an absorber to absorb X-ray photons and convert to heat, a thermometer to measure the temperature rise of the absorber, and a thermal link to a heat sink to allow the absorbed heat to escape from the absorber (Figure 2.4). The temperature rise can be given by

$$\Delta T = \frac{E}{C},\tag{2.1}$$

where E is the energy of incident X-ray photon, and C is the heat capacity of the detector. The heat converted from the energy in the absorber escapes to the heat sink, and the detector gradually returns back to the equilibrium point. This thermal response can be given by

$$C\frac{d\Delta T}{dt} = -G\Delta T,\tag{2.2}$$

where G is the thermal conductivity of the link to the sink. Therefore, the temperature of the detector falls exponentially with a time constant given by

$$\tau_0 = \frac{C}{G}.\tag{2.3}$$

The energy resolution of a X-ray microcalorimeter is limited by a thermal fluctuation in the detector. The number of phonons in a detector is $N \sim CT/k_{\rm B}T = C/k_{\rm B}$, and the thermal fluctuation of the detector will then be

$$\Delta U \sim \sqrt{N} k_{\rm B} T = \sqrt{k_{\rm B} T^2 C}.$$
(2.4)



Fig. 2.5 Transition edge

Generally the intrinsic energy resolution of a X-ray microcalorimeter is

$$\Delta E_{\rm FWHM} = 2.35\xi \sqrt{k_{\rm B}T^2C},\tag{2.5}$$

as we derive in Section 2.2.5. ξ is a parameter determined by a thermometer sensitivity and the like.

2.2.2 Transition Edge Sensors

Thermometers used for measuring the temperature rise of microcalorimeters are usually resistance thermometers. The sensitivity of thermometer α is defined as

$$\alpha \equiv \frac{d\log R}{d\log T} = \frac{T}{R} \frac{dR}{dT},$$
(2.6)

where T is temperature of the thermometer, and R is resistance of the thermometer. Historically, semiconductor thermistors have been used for microcalorimeters; however, the sensitivity of semiconductor thermistors may not be very high. For instance, a sensitivity of semiconductor thermistor used for Suzaku's X-ray microcalorimeter (XRS) is $|\alpha| \sim 6$.

A Transition Edge Sensor (TES) is a thermometer to archive a very high sensitivity (typically $|\alpha| > 100$) utilizing a very steep resistivity change at a transition edge between its super conducting and normal state. The range of the transition is typically a few mK (Figure 2.5), which makes the sensitivity even higher than 1000. This high sensitivity allows a better energy resolution by more than a order of magnitude if compared with microcalorimeters using semiconductor thermistors. It also allows more selectability of absorber materials and shapes.

As the TES uses the transition edge, the operating temperature of TES microcalorimeter is limited to the transition temperature. By making a TES bilayer with a normal metal such as AU, the transition temperature can however be controlled using a proximity effect.



Fig. 2.6 Constant voltage biasing (left), and constant voltage biasing using a shunt resistor (right)

2.2.3 Electrothermal Feedback (ETF)

TES may have the very high sensitivity, but the temperature range that keeps the high sensitivity is quite small (\sim mK), and therefore the operating point needs to be kept within the transition edge. It can be realized by biasing TES with a constant voltage and applying a strong negative feedback, called an Electro-Thermal Feedback (ETF) [17].

Suppose a TES is biased with a constant voltage as shown in Figure 2.6 (left). The resistance of the TES rapidly increases as the temperature rises due to an incident photon, making the current flow as well as the Joule heat smaller under the voltage bias. The operating temperature is then kept in a stable. In reality constant voltage biasing is realized by using a shunt resistor inserted in parallel with the TES due to a non-negligible resistance in wirings between room temperature and a cryostage (Figure 2.6 right). For now we assume ideal constant voltage biasing.

Thermal conductivity is defined as

$$G \equiv dP/dT.$$
(2.7)

Generally thermal conductivity has a temperature dependence and G is represented as

$$G = G_0 T^{n-1}.$$
 (2.8)

When a heat is carried by electrons, then n = 2, and when it is carried by phonons, then n = 4.

Let us consider a thermal conductivity between a TES and a heat sink (heat bath). A heat flow from the TES to the sink can be given by

$$P = \int_{T_{\text{bath}}}^{T} G dT = \frac{G_0}{n} \left(T^n - T_{\text{bath}}^n \right),$$
(2.9)

by integrating (2.7).

In thermal equilibrium the Joule heating of the TES, $P_{\rm b} \equiv V_{\rm b}^2/R_0$, and the heat flow from the TES to the sink is balanced, then gives

$$P_{\rm b} = \frac{G_0}{n} \left(T_0^n - T_{\rm bath}^n \right), \tag{2.10}$$

where T_0 is the TES temperature, V_b is the bias voltage, R_0 is the TES resistance, and T_{bath} is the temperature of the heat sink.

When the TES temperature becomes T with a small temperature rise $\Delta T \equiv T - T_0$, the change of internal energy is the same as the change of heat, so

$$C\frac{dT}{dt} = \frac{V_{\rm b}^2}{R(T)} - \frac{G_0}{n} \left(T^n - T_{\rm bath}^n\right)$$
(2.11)

is given. In the first order approximation, (2.11) becomes

$$C\frac{d\Delta T}{dt} \simeq -\frac{V_{\rm b}^2}{R_0^2}\Delta R - G_0 T^{n-1}\Delta T$$
(2.12)

$$=\frac{P_{\rm b}\alpha}{T}\Delta T - G\Delta T.$$
(2.13)

G is the thermal conductivity of TES at the temperature T. The solution of (2.12) is

$$\Delta T = \Delta T_0 \exp\left(-\frac{t}{\tau_{\text{eff}}}\right),\tag{2.14}$$

where

$$\tau_{\rm eff} \equiv \frac{C/G}{1 + \frac{P_{\rm b}\alpha}{GT}} \tag{2.15}$$

$$=\frac{\tau_0}{1+\frac{P_{\rm b}\alpha}{GT}}\tag{2.16}$$

is an effective time constant. With (2.10) and (2.16), $\tau_{\rm eff}$ can be expressed as

$$\tau_{\rm eff} = \frac{\tau_0}{1 + \frac{\alpha}{n} \left(1 - \left(\frac{T_{\rm bath}}{T}\right)^n\right)}.$$
(2.17)

Moreover, when the temperature of the sink is far less than the temperature of TES $(T_{\text{bath}}^n \ll T^n)$, τ_{eff} can be approximated as

$$\tau_{\rm eff} = \frac{\tau_0}{1 + \frac{\alpha}{n}} \tag{2.18}$$

$$\approx \frac{n}{\alpha} \tau_0.$$
 (2.19)

(2.19) is, however, only when $\alpha/n \gg 1$. As seen in this equation, the TES response time is significantly shortened due to the ETF when α is large. The energy of incident X-ray photon can be sensed as a current change

$$\Delta I = \frac{V_{\rm b}}{R(T_0 + \Delta T)} - \frac{V_{\rm b}}{R(T_0)}$$
(2.20)

$$\simeq -\frac{\Delta R}{R}I\tag{2.21}$$

$$\simeq -\alpha \frac{E}{CT}I.$$
(2.22)

General Formalization of ETF and Current Responsivity

Let us consider the response of the voltage-biased microcalorimeter to an externally applied time-dependent microscopic power $\delta P e^{i\omega t}$. The response is assumed to be linear, and the temperature change can then be given by $\delta T e^{i\omega t}$. Without the feedback, we have

$$P_{\rm bgd} + \delta P e^{i\omega t} = \bar{G}(T - T_{\rm bath}) + G\delta T e^{i\omega t} + i\omega C\delta T e^{i\omega t}, \qquad (2.23)$$

where P_{bgd} is a background power, and \bar{G} is an averaged thermal conductivity. In thermal equilibrium, the background power is

$$P_{\text{bgd}} = \bar{G}(T - T_{\text{bath}}). \tag{2.24}$$

With (2.23) and (2.24), δT is

$$\delta T = \frac{1}{G} \frac{1}{1 + i\omega\tau_0} \delta P, \qquad (2.25)$$

using δP , where $\tau_0 \equiv C/G$ is the intrinsic time constant.

With the ETF, the equation of energy conservation is given by

$$P_{\rm bgd} + \delta P e^{i\omega t} + P_{\rm b} + \delta P_{\rm b} e^{i\omega t} = \bar{G}(T - T_{\rm bath}) + G\delta T e^{i\omega t} + i\omega C\delta T e^{i\omega t}.$$
(2.26)

With voltage biasing, we have

$$\delta P_{\rm b} e^{i\omega t} = \frac{dP_{\rm b}}{dI} \delta I e^{i\omega t} = V_{\rm b} \delta I e^{i\omega t}, \qquad (2.27)$$

$$\delta I e^{i\omega t} = \frac{dI}{dR} \delta R e^{i\omega t} = \frac{d}{dR} \left(\frac{V_{\rm b}}{R} \right) \delta R e^{i\omega t} = -\frac{V_{\rm b}}{R^2} \delta R e^{i\omega t}, \qquad (2.28)$$

$$\delta R e^{i\omega t} = \frac{dR}{dT} \delta T e^{i\omega t} = \alpha \frac{R}{T} \delta T e^{i\omega t}.$$
(2.29)

Using these, (2.26) becomes

$$P_{\rm bgd} + \delta P e^{i\omega t} + \frac{V_b^2}{R} - \frac{V_b^2}{R^2} \frac{dR}{dT} \delta T e^{i\omega t} = \bar{G}(T - T_{\rm bath}) + G\delta T e^{i\omega t} + i\omega C\delta T e^{i\omega t}.$$
 (2.30)

The solution to (2.30) is then

$$\delta T e^{i\omega t} = \frac{1}{\alpha \frac{P_{\rm b}}{\sigma} + G + i\omega C} \delta P e^{i\omega t}$$
(2.31)

$$= \frac{1}{G} \frac{1}{1 + \frac{\alpha P_{\rm b}}{GT}} \frac{1}{1 + i\omega \tau_{\rm eff}} \delta P \mathrm{e}^{i\omega t}, \qquad (2.32)$$



Fig. 2.7 Electrothermal feedback

where

$$\tau_{\rm eff} \equiv \frac{1}{1 + \frac{\alpha P_{\rm b}}{GT}} \frac{C}{G}$$
(2.33)

is the effective time constant under the ETF.

According to the general feedback theory, the ETF system can be described as Figure 2.7. The feedback b and the loopgain $\mathcal{L}(\omega)$ are

$$b = -V_{\rm b},\tag{2.34}$$

$$\mathcal{L}(\omega) = \frac{1}{G(1+i\omega\tau_0)} \times \alpha \frac{R}{T} \times \left(-\frac{I}{R}\right) \times (-V_{\rm b}) = \frac{\alpha P_{\rm b}}{GT} \frac{1}{1+i\omega\tau_0} \equiv \frac{\mathcal{L}_0}{1+i\omega\tau_0},\tag{2.35}$$

where

$$\mathcal{L}_0 \equiv \frac{\alpha P_{\rm b}}{GT} \tag{2.36}$$

is the loopgain at DC. The closed-loop transfer function, or the current responsivity

$$S_I(\omega) \equiv \frac{\delta I}{\delta P} \tag{2.37}$$

can be expressed as

$$S_I(\omega) = \frac{1}{b} \frac{\mathcal{L}(\omega)}{1 + \mathcal{L}(\omega)}$$
(2.38)

$$= -\frac{1}{V_{\rm b}} \frac{\mathcal{L}_0}{\mathcal{L}_0 + 1 + i\omega\tau_0} \tag{2.39}$$

$$= -\frac{1}{V_{\rm b}} \frac{\mathcal{L}_0}{\mathcal{L}_0 + 1} \frac{1}{1 + i\omega\tau_{\rm eff}}$$
(2.40)

using $\mathcal{L}(\omega)$, where

$$\tau_{\rm eff} \equiv \frac{\tau}{\mathcal{L}_0 + 1}.\tag{2.41}$$

With a large loopgain $(\mathcal{L}_0 \gg 1)$, the transfer function becomes

$$S_I(\omega) = -\frac{1}{V_b} \frac{1}{1 + i\omega\tau_{\text{eff}}}.$$
 (2.42)

In the frequency range $\omega \ll 1/\tau_{\rm eff}$ we have

$$S_I = -\frac{1}{V_{\rm b}}.$$
 (2.43)

A response to an input $P(t) = E\delta(t)$ can be calculated as

$$P(\omega) = \frac{1}{2\pi} \int_{-\infty}^{\infty} E\delta(t) e^{i\omega t} dt$$
(2.44)

$$=\frac{E}{2\pi},\tag{2.45}$$

and the output becomes

$$I(\omega) = S_I(\omega)P(\omega) \tag{2.46}$$

$$= -\frac{E}{2\pi V_{\rm b}} \frac{\mathcal{L}_0}{\mathcal{L}_0 + 1} \frac{1}{1 + i\omega \tau_{\rm eff}}.$$
 (2.47)

By an inverse Fourier transform, the current output in the time domain becomes

$$I(t) = \int_{-\infty}^{\infty} I(\omega) e^{-i\omega t} d\omega$$
(2.48)

$$= -\frac{1}{2\pi} \frac{E}{V_{\rm b}} \frac{\mathcal{L}_0}{\mathcal{L}_0 + 1} \int_{-\infty}^{\infty} \frac{\mathrm{e}^{-i\omega t}}{1 + i\omega\tau_{\rm eff}} d\omega$$
(2.49)

$$= -\frac{E}{V_{\rm b}\tau_{\rm eff}} \frac{\mathcal{L}_0}{\mathcal{L}_0 + 1} \exp\left(-\frac{t}{\tau_{\rm eff}}\right)$$
(2.50)

$$= -\frac{\alpha E}{CT} I_0 \exp\left(-\frac{t}{\tau_{\text{eff}}}\right),\tag{2.51}$$

where I_0 is the current through the TES in the steady state. On the other hand, the temperature rise due to the input $P(t) = E\delta(t)$ can be expressed as

$$\Delta T(\omega) = \frac{1}{G(1+i\omega\tau_0)} \frac{1}{1+\mathcal{L}(\omega)} P(\omega)$$
(2.52)

$$= \frac{1}{2\pi} \frac{E}{G} \frac{1}{1 + \mathcal{L}_0} \frac{1}{1 + i\omega\tau_{\text{eff}}}$$
(2.53)

in the frequency domain, and converting back to the time domain we have

$$\Delta T(t) = \int_{-\infty}^{\infty} \Delta T(\omega) \mathrm{e}^{-i\omega t} d\omega$$
(2.54)

$$= \frac{1}{2\pi} \frac{E}{G} \frac{1}{\mathcal{L}_0 + 1} \int_{-\infty}^{\infty} \frac{\mathrm{e}^{-i\omega t}}{1 + i\omega\tau_{\mathrm{eff}}} d\omega$$
(2.55)

$$= \frac{E}{G\tau_{\rm eff}} \frac{1}{\mathcal{L}_0 + 1} \exp\left(-\frac{t}{\tau_{\rm eff}}\right)$$
(2.56)

$$= \frac{E}{C} \exp\left(-\frac{t}{\tau_{\text{eff}}}\right). \tag{2.57}$$

When \mathcal{L}_0 is considered to be constant, (2.50) gives

$$\int V_{\rm b}I(t)dt = -\frac{\mathcal{L}_0}{\mathcal{L}_0 + 1}E.$$
(2.58)

Thus, the integration of Joule heating due to an incident X-ray photon is proportional to the incident energy E. For the energy, $\mathcal{L}_0/(\mathcal{L}_0 + 1)$ is compensated by the change of Joule heating, and $1/(\mathcal{L}_0 + 1)$ escapes to the heat sink. When $\mathcal{L}_0 \gg 1$, the integration of Joule heating is equivalent to the incident energy.



Fig. 2.8 Electrothermal feedback with intrinsic noises

2.2.4 Intrinsic Noises

The essential noise of TES is Johnson noise (thermal noise) and phonon noise (thermal fluctuation noise), and the energy resolution of TES is limited to these noises.

TES suffers from two intrinsic noises. One is Johnson noise arisen from the TES resistance, and the other is phonon noise arisen from the exchange of energy between the TES and the heat sink. Figure 2.8 shows the ETF diagram with these intrinsic noises. Because of their origins, the positions where those noises are input are slightly different. The current fluctuation due to a small thermal fluctuation $\delta P_{\rm ph}$ is

$$\delta I_{\rm ph} = -\frac{1}{V_{\rm b}} \frac{\mathcal{L}(\omega)}{1 + \mathcal{L}(\omega)} \delta P_{\rm ph}$$
(2.59)

$$=S_I \delta P_{\rm ph}.\tag{2.60}$$

Thus, the current spectral density of phonon noise is given by

$$\delta I_{\rm ph}^2 = |S_I|^2 \delta P_{\rm ph}^2 \tag{2.61}$$

$$= \frac{1}{V_{\rm b}^2} \left(\frac{\mathcal{L}_0}{\mathcal{L}_0 + 1}\right)^2 \frac{1}{1 + \omega^2 \tau_{\rm eff}^2} \delta P_{\rm ph}^2.$$
(2.62)

According to Mather [21], the power spectral density of phonon noise in $0 \le f < \infty$ can be given by

$$\delta P_n^2 = 4k_B G T^2 \frac{\int_{T_{\text{bath}}}^T \left(\frac{t\kappa(t)}{T\kappa(T)}\right)^2 dt}{\int_{T_{\text{tot}}}^T \left(\frac{\kappa(t)}{\kappa(T)}\right) dt}$$
(2.63)

$$\equiv 4k_B G T^2 \Gamma, \tag{2.64}$$

where $\kappa(T)$ is a thermal conductivity of thermal link. Assuming $\theta \equiv T_{\text{bath}}/T$ and $\kappa(T) = \kappa(T_{\text{bath}})\theta^{-(n-1)}$, Γ can be written as

$$\Gamma = \frac{n}{2n+1} \frac{1 - \theta^{(2n+1)}}{1 - \theta^n}.$$
(2.65)

Using (2.64) and (2.62), the current spectral density of phonon noise can be expressed as

$$\delta I_{\rm ph}^2 = 4k_{\rm B}GT^2\Gamma|S_I|^2 \tag{2.66}$$

$$= \frac{4k_{\rm B}GT^2\Gamma}{b^2} \left(\frac{\mathcal{L}_0}{\mathcal{L}_0 + 1}\right)^2 \frac{1}{1 + \omega^2 \tau_{\rm eff}^2} \tag{2.67}$$

$$= \frac{4k_{\rm B}GT^2\Gamma}{V_{\rm b}^2} \left(\frac{\mathcal{L}_0}{\mathcal{L}_0 + 1}\right)^2 \frac{1}{1 + \omega^2 \tau_{\rm eff}^2}.$$
 (2.68)

On the other hand, the current fluctuation $\delta I_{\rm J}^0$ due to Johnson noise $\delta V_{\rm J}$ is

$$\delta I_{\rm J}^0 = \frac{\delta V_{\rm J}}{R},\tag{2.69}$$

and the output fluctuation becomes

$$\delta I_{\rm J} = \frac{1}{1 + \mathcal{L}(\omega)} \delta I_{\rm J}^0 \tag{2.70}$$

$$=\frac{\frac{1}{\mathcal{L}_0+1}+i\omega\tau_{\text{eff}}}{1+i\omega\tau_{\text{eff}}}\frac{\delta V_{\text{J}}}{R}$$
(2.71)

$$=\frac{1}{\mathcal{L}_0+1}\frac{1+i\omega\tau_0}{1+i\omega\tau_{\rm eff}}\frac{\delta V_{\rm J}}{R}.$$
(2.72)

The current spectral density of Johnson noise is given by $\delta V_{\rm J}^2 = 4k_{\rm B}RT$ in $0 \leq f < \infty$, the output current spectral density then becomes

$$\delta I_{\rm J}^2 = \frac{4k_{\rm B}T}{R} \left(\frac{1}{\mathcal{L}_0 + 1}\right)^2 \left|\frac{1 + i\omega\tau_0}{1 + i\omega\tau_{\rm eff}}\right|^2 \tag{2.73}$$

$$= \frac{4k_{\rm B}T}{R} \left(\frac{1}{\mathcal{L}_0 + 1}\right)^2 \frac{1 + \omega^2 \tau_0^2}{1 + \omega^2 \tau_{\rm eff}^2}$$
(2.74)

$$= \begin{cases} \frac{4k_{\rm B}T}{R} \left(\frac{1}{\mathcal{L}_0 + 1}\right)^2 & \text{if } \omega \ll \tau_0^{-1} \\ \frac{4k_{\rm B}T}{R} & \text{if } \omega \gg \tau_{\rm eff}^{-1} \end{cases}$$
(2.75)

Note that in the frequency range $\omega \ll \tau_0^{-1}$, the Johnson noise is suppressed by the ETF, while in the frequency range $\omega \gg \tau_{\text{eff}}^{-1}$, it becomes the original value.

The total noise current spectral density is given by the sum of squares of the phonon noise and the Johnson noise as

$$\delta I^2 = \delta I_{\rm J}^2 + \delta I_{\rm ph}^2 \tag{2.76}$$

$$= \frac{4k_{\rm B}T}{R} \left(\frac{1}{\mathcal{L}_0 + 1}\right)^2 \frac{1 + \omega^2 \tau_0^2}{1 + \omega^2 \tau_{\rm eff}^2} + 4k_{\rm B}GT^2 \Gamma \frac{1}{V_{\rm b}^2} \left(\frac{\mathcal{L}_0}{\mathcal{L}_0 + 1}\right)^2 \frac{1}{1 + \omega^2 \tau_{\rm eff}^2}$$
(2.77)
$$1 + \Gamma \alpha \mathcal{L}_0 = 2^{-2} 2^{-2}$$

$$=\frac{4k_{\rm B}T}{R}\frac{\frac{1+13\omega_0}{(\mathcal{L}_0+1)^2}+\omega^2\tau_{\rm eff}^2}{1+\omega^2\tau_{\rm eff}^2}$$
(2.78)

in $0 \le f < \infty$. In the limit of strong ETF, it becomes

$$\delta I^{2} = \frac{4k_{\rm B}T}{R} \frac{n/2 + \omega^{2} \tau_{\rm eff}^{2}}{1 + \omega^{2} \tau_{\rm eff}^{2}}.$$
(2.79)



Fig. 2.9 Examples of noise current power spectral density. The left plot shows when $\alpha = 100$, while the right shows when $\alpha = 1000$.

Figure 2.9 shows the frequency dependencies of the noise current power spectral density. To see the relation of the phonon noise and Johnson noise, we have the ratio of those two terms as

$$\frac{\delta I_{\rm ph}^2}{\delta I_{\rm I}^2} = \frac{\alpha \mathcal{L}_0 \Gamma}{1 + \omega^2 \tau_0^2}.$$
(2.80)

Thus, in the low frequency region the Johnson noise is suppressed and the phonon noise is $\alpha \mathcal{L}_0 \Gamma$ time larger, but in the region $\omega > \tau_0^{-1}$ the Johnson noise starts dominating, and in the region $\omega \gg \tau_{\text{eff}}^{-1}$ the Johnson noise finally dominates.

In contrast, the ratio of the signal and phonon noise is frequency independent as

$$\frac{\delta P_{\text{signal}}^2}{\delta P_{\text{n}}} = \frac{2E^2}{4k_B G T^2 \Gamma},\tag{2.81}$$

because those two have identical frequency dependencies.

With (2.40) and (2.75), the Johnson noise can be expressed as

$$\delta I_{\rm J}^2 = \frac{4k_{\rm B}T}{R} \frac{b^2(1+\omega^2\tau_0^2)}{\mathcal{L}_0^2} |S_I|^2 \tag{2.82}$$

using the current responsivity S_I . From (2.67) and (2.75), the intrinsic noise is given by

$$\delta I^2 = \frac{4k_B T}{R} \frac{1 + \omega^2 \tau_0^2}{\mathcal{L}_0^2} b^2 |S_I|^2 + 4k_B G T^2 \Gamma |S_I|^2.$$
(2.83)

The noise equivalent power NEP(f) is defined as

$$\operatorname{NEP}(f)^2 = \left|\frac{\delta I}{S_I}\right|^2,\tag{2.84}$$

and the NEP(f) for the intrinsic noise can been calculated as

$$\operatorname{NEP}(f)^2 = \left|\frac{\delta I}{S_I}\right|^2 \tag{2.85}$$

$$=\frac{4k_BT}{R}\frac{b^2}{\mathcal{L}_0^2}\left(1+(2\pi f)^2\tau_0^2+\frac{\mathcal{L}_0^2}{b^2}RGT\Gamma\right)$$
(2.86)

$$=4k_BTP_{\rm b}\left(\frac{1+(2\pi f)^2\tau_0^2}{\mathcal{L}_0^2}+\frac{\alpha\Gamma}{\mathcal{L}_0}\right).$$
(2.87)

2.2.5 Energy Resolution

Once the NEP is given, the energy resolution is given by

$$\Delta E_{\rm rms} = \left(\int_0^\infty \frac{4df}{\rm NEP^2(f)}\right)^{-\frac{1}{2}} \tag{2.88}$$

according to Moseley et al. [22]. Using (2.87), the energy resolution for the intrinsic noises is thus

$$\Delta E_{\rm rms} = \left(\int_0^\infty \frac{4df}{\frac{4k_B T}{R} \frac{b^2}{\mathcal{L}_0^2} \left((1 + (2\pi f)^2 \tau_0^2) + \frac{\mathcal{L}_0^2}{b^2} RGT\Gamma \right)} \right)^{-\frac{1}{2}}$$
(2.89)

$$=\sqrt{\frac{4k_{\rm B}T}{R}\frac{b^2}{\mathcal{L}_0^2}\tau_0}\sqrt{1+\frac{\mathcal{L}_0^2}{b^2}RGT\Gamma}$$
(2.90)

$$= \sqrt{4k_{\rm B}T^2 C \frac{b^2}{RGT\mathcal{L}_0^2} \sqrt{1 + \frac{\mathcal{L}_0^2}{b^2} RGT\Gamma}}.$$
(2.91)

Defining ξ as

$$\xi \equiv 2 \sqrt{\frac{b^2}{RGT\mathcal{L}_0^2}} \sqrt{1 + \frac{\Gamma}{\frac{b^2}{RGT\mathcal{L}_0^2}}},$$
(2.92)

we have the full width at half maximum (FWHM) intrinsic energy resolution as

$$\Delta E_{\rm FWHM} = 2.35\xi \sqrt{k_{\rm B}T^2C}.$$
(2.93)

From (2.34) and (2.36), (2.92) becomes

$$\xi = 2\sqrt{\frac{1}{\alpha\mathcal{L}_0}\sqrt{1+\alpha\mathcal{L}_0\Gamma}}.$$
(2.94)

2.2.6 Optimal Filtering

An acquired X-ray pulse signal is usually contaminated by noises, and taking its peak value as a pulse height does not achieve the best energy resolution. In this type of application, optimal filtering^{*1} is generally used to retrieve the pulse height minimizing the deterioration due to noises.

^{*1} The term optimal filtering sometimes refers the Wiener filter, but here it refers the matched filter.

Let us assume D(t) to be the acquired pulse signal, and in frequency domain it can be represented as

$$D(f) = A \times M(f) + N(f), \qquad (2.95)$$

where M(f) and N(f) are ideal spectra of the pulse and the noise, and A is the amplitude of the pulse. M(f) is also called a model pulse. We now determine A so that the difference of the acquired pulse and the model pulse is minimized using the least-square method. Defining the difference of the acquired pulse and the model pulse as

$$\chi^2 \equiv \int \frac{|D(f) - A \times M(f)|^2}{|N(f)|^2},$$
(2.96)

A that minimize χ^2 is given by

$$A = \frac{\int_{-\infty}^{\infty} \frac{DM^* + D^*M}{2|N|^2} df}{\int_{-\infty}^{\infty} \frac{|M|^2}{|N|^2} df}.$$
(2.97)

D(f) and M(f) are Fourier transformations of real functions, and thus $D(-f) = D(f)^*$ and $M(-f) = M(f)^*$. We therefore have

$$\int_{-\infty}^{\infty} \frac{D(f)M(f)^*}{2|N|^2} df = -\int_{\infty}^{-\infty} \frac{D(-f)M(-f)^*}{2|N|^2} df = \int_{-\infty}^{\infty} \frac{M(f)D(f)^*}{2|N|^2} df,$$
(2.98)

and A can be given as

$$A = \frac{\int_{-\infty}^{\infty} \frac{DM^*}{|N|^2} df}{\int_{-\infty}^{\infty} \frac{|M|^2}{|N|^2} df},$$
(2.99)

or

$$A = \frac{\int_{-\infty}^{\infty} \frac{D}{M} \left| \frac{M}{N} \right|^2 df}{\int_{-\infty}^{\infty} \left| \frac{M}{N} \right|^2 df}.$$
(2.100)

According to (2.100), A is the weighted average of D(f)/M(f) with the weight of $[M(f)/N(f)]^2$, which is the signal-to-noise ratio. (2.100) can be further transformed into

$$A = \frac{\int_{-\infty}^{\infty} D(t)\mathcal{F}^{-1}\left(\frac{M(f)}{|N(f)|^2}\right)dt}{\int_{-\infty}^{\infty} \left|\frac{M}{N}\right|^2 df},$$
(2.101)

where \mathcal{F}^{-1} is the inverse-Fourier transformation. We now define a template of optimal filtering T(t) as

$$T(t) \equiv \mathcal{F}^{-1}\left(\frac{M(f)}{|N(f)|^2}\right).$$
(2.102)

Using (2.102), the pulse height H is given by

$$H = N \int_{-\infty}^{\infty} D(t)T(t)dt, \qquad (2.103)$$

or for discrete signal data

$$H = N \sum_{i} D_i(t) T_i(t),$$
 (2.104)

where N is a normalized constant, and $D_i(t)$ and $T_i(t)$ are the quantized pulse and template.

To calculate T(t), we need M(f) and N(f). They are generally estimated from the ensemble averages of the pulse and the noise as

$$M(t) = \langle D(t) \rangle \tag{2.105}$$

and

$$N(f) = \langle E(f) \rangle, \qquad (2.106)$$

where E(f) is the acquired noise signal in the frequency domain.



Fig. 2.10 RCSJ model

2.3 Superconducting Quantum Interference Devices

2.3.1 Josephson Junctions

A Josephson junctions consists of two superconductors separated by a weak insulator. If the coupling is weak enough, a super-current $I_{\rm S}$ flows between the superconductors. Let δ be the phase difference across the superconductors, the super-current $I_{\rm S}$ is then

$$I_{\rm S} = I_0 \sin \delta. \tag{2.107}$$

 I_0 is called a critical current of the junction, and gives the maximum current of the super-current. No voltage drop thus occurs at the junction when the current through the junction is below I_0 .

When the phase difference δ evolves with time t, a voltage U is generated across the junction as

$$\dot{\delta} \equiv \frac{d\delta}{dt} = \frac{2e}{\hbar}U = \frac{2\pi}{\Phi_0}U,\tag{2.108}$$

where $\Phi_0 = h/2e \approx 2.07 \times 10^{-15}$ Wb is called the flux quantum.

2.3.2 RCSJ Model

Josephson junctions are generally modeled by a resistively- and capacitively-shunted junction (RCSJ) model. Figure 2.10 shows the equivalent circuit of RCSJ model. From Kirchhoff's laws, we have

$$C\dot{U} + \frac{U}{R} + I_0 \sin \delta = I + I_N(t).$$
 (2.109)

For now we ignore the noise term $I_{\rm N}(t)$, and (2.109) then becomes

$$\frac{\Phi_0}{2\pi}C\ddot{\delta} + \frac{\Phi_0}{2\pi}\frac{1}{R}\dot{\delta} = I - I_0\sin\delta = -\frac{2\pi}{\Phi_0}\frac{\partial U_J}{\partial\delta}$$
(2.110)

using (2.108). $U_{\rm J}$ is called the tilted washboard potential and given by

$$U_{\rm J} \equiv \frac{\Phi_0}{2\pi} \left\{ I_0(1 - \cos\delta) - I\delta \right\} = E_{\rm J}(1 - \cos\delta - i\delta), \qquad (2.111)$$

where E_J is the Josephson coupling energy given by $E_J = I_0 \Phi_0/2\pi$, and *i* is the normalized bias current given by $i \equiv I/I_0$. U_J can be normalized using E_J and let u_J be

$$u_{\rm J} = 1 - \cos \delta - i\delta. \tag{2.112}$$

When $I < I_0$, the phase is trapped in one of the minima of the washboard potential, and the response of the junction then oscillates with the plasma frequency ω_p , which is given by

$$\omega_{\rm p,i} = \omega_{\rm p} (1 - i^2)^{1/4} \quad \text{for} \quad \omega_{\rm p} = \left(\frac{2\pi}{\Phi_0} \frac{I_0}{C}\right)^{1/2}.$$
(2.113)

In this case the time average is zero, so the voltage across the junction is also zero. When $I > I_0$, the phase can move over potential walls, and $\dot{\delta}$ and the voltage across the junction then becomes non-zero. Substituting (2.110) with the characteristic frequency $\omega_c \equiv 2\pi I_0 R/\Phi_0$, which is the Josephson frequency at the characteristic voltage $V_c \equiv I_0 R$, we have

$$\frac{\ddot{\delta}}{\omega_{\rm p}^2} + \frac{\dot{\delta}}{\omega_{\rm c}} = i - \sin \delta = -\frac{\partial u_{\rm J}}{\partial \delta},\tag{2.114}$$

or

$$\beta_{\rm C} \frac{\ddot{\delta}}{\omega_{\rm c}^2} + \frac{\dot{\delta}}{\omega_{\rm c}} = i - \sin \delta = -\frac{\partial u_{\rm J}}{\partial \delta}.$$
(2.115)

 $\beta_{\rm C}$ is called Stewart-McCumber and defined as

$$\beta_{\rm C} \equiv \left(\frac{\omega_{\rm c}}{\omega_{\rm p}}\right)^2 = \frac{2\pi}{\Phi_0} I_0 R^2 C.$$
(2.116)

Let us consider two limits, $\beta_{\rm C} \ll 1$ and $\beta_{\rm C} \gg 1$, for (2.115).

In the limit of $\beta_{\rm C} \ll 1$, which is called the over-damped limit, we can ignore the inertial term in (2.115), which is comparable to the negligible junction capacitance. In this limit the RCSJ model is simplified to the RSJ model. Since the inertial term is negligible, the particle gets trapped in the potential at $I = I_0$ when reducing the bias current I, and the voltage across the junction instantly becomes zero. The I-V characteristic is therefore non-hysteretic. Assuming $\beta_{\rm C} = 0$, we can solve (2.115) for the normalized voltage $u \equiv U/I_0 R$ as

$$u(t) = \begin{cases} 0 & i < 1\\ \frac{i^2 - 1}{i + \cos \omega t} & i > 1 \end{cases},$$
(2.117)

where $\omega = \omega_c \sqrt{i^2 - 1}$. When i > 1, v oscillates with the frequency ω , and the frequency increases as i increases. The normalized time averaged voltage $v \equiv V/I_0 R$ (V is time averaged voltage) is

$$v = \begin{cases} 0 & i < 1\\ \sqrt{i^2 - 1} & i > 1 \end{cases}.$$
 (2.118)

In the limit of $\beta_{\rm C} \gg 1$, which is called the under-damped limit, the dynamics of junction is determined by the RC circuit since the Josephson frequency is much higher than the relaxation frequency ω_{RC} . The voltage across the junction keeps a finite value even at $I < I_0$ (or i < 1) when reducing I. On the other hand, when increasing I the voltage across the junction stays zero until $I > I_0$ (or i > 1). In this limit the I-V characteristic is severely



Fig. 2.11 Calculated hysteretic I-V characteristics for various values of $\beta_{\rm C}$ (left) and the return critical current vs. $\beta_{\rm C}$ (right). Adapted from Clarke and Braginski [6].

hysteretic.

Figure 2.11 (left) shows the calculated I-V characteristics for various values of $\beta_{\rm C}$ [6]. Reducing the bias current I from above I_0 , the voltage across the junction becomes zero at the return critical current $I_{\rm r}(\beta_{\rm C})$, which is the function of $\beta_{\rm C}$. The normalized return critical current $i_{\rm r} = I_{\rm r}/I_0$ is also the function of $\beta_{\rm C}$, and it monotonically decreases from 1 to 0 for increasing $\beta_{\rm C}$. Figure 2.11 (right) shows the approximated analytical solutions as well as the result of numerical simulation [6].

To achieve non-hysteretic I-V characteristic, a shunt resistor is generally inserted in parallel with the junction to minimize $\beta_{\rm C}$ below 1.

2.3.3 Johnson-Nyquist Noise

The current fluctuation $I_{\rm N}$ due to Johnson noise is

$$I_{\rm N}^2 = \frac{4k_{\rm B}T}{R}.$$
 (2.119)

Let $i_N \equiv I_N/I_0$ be the normalized noise current, the washboard potential (2.111) now becomes

$$U_{\rm J,N} = E_{\rm J} \left\{ 1 - \cos \delta - [i + i_{\rm N}(t)] \delta \right\}, \qquad (2.120)$$

which means that the averaged tilt of the potential fluctuate with the normalized noise current.

Here we define a noise parameter Γ as the ratio of the thermal energy over the Josephson coupling energy

$$\Gamma \equiv \frac{k_{\rm B}T}{E_{\rm J}} = \frac{2\pi k_{\rm B}T}{I_0 \Phi_0}.$$
(2.121)

Using Γ and $\omega_{\rm c}$, $i_{\rm N}$ now becomes

$$i_{\rm N}^2 = \frac{4\Gamma}{\omega_{\rm c}}.$$

Even when $I < I_0$, the total current through the junction with noise, $I + I_N(t)$, sometime exceeds the critical current I_0 . When it happens, the phase move over the next minima in the potential, which makes a pulse-like voltage across the junction. In this case the time-averaged voltage becomes a finite value, making a rounding effect at the transition edge of superconducting state and normal state. Figure 2.12 shows the I-V characteristic obtained by numerically solving

$$\beta_{\rm C} \frac{\ddot{\delta}}{\omega_{\rm c}^2} + \frac{\dot{\delta}}{\omega_{\rm c}} = i + i_{\rm N}(t) - \sin\delta \tag{2.122}$$


Fig. 2.12 Calculated I-V characteristic for various values of Γ for negligible capacitance (left) and finite capacitance (right). Adapted from Clarke and Braginski [6].



Fig. 2.13 The dc SQUID: the schematic diagram of dc SQUID (left) and the equivalent circuit of dc SQUID

[6]. Figure 2.12 (left) shows the rounding effect due to Γ . The hysteresis in the I-V characteristic in $\beta_{\rm C} > 1$ is also minimized by Γ . This is because $I + I_{\rm N}(t)$ can be less than I_r even when $I > I_r$ due to the current fluctuation.

From the definition of Γ we see the increase of noise when the operating temperature T becomes larger or the critical current I_0 becomes smaller. If we redefine Γ as $\Gamma = I_{\rm th}/I_0$, the thermal noise current $I_{\rm th}$ is defined by

$$I_{\rm th} \equiv \Gamma I_0 = \frac{2\pi}{\Phi_0} k_{\rm B} T. \qquad (2.123)$$

Practically Γ should be less than one, and $I_{\rm th}$ therefore gives the minimum critical current. At liquid nitrogen temperature, $I_{\rm th} = 3.23 \,\mu\text{A}$, and at liquid helium temperature, $I_{\rm th} = 176 \,\text{nA}$. The critical current should be less than those.

2.3.4 dc SQUID

A dc SQUID consists of two Josephson junctions and a superconducting ring. Figure 2.13 shows the schematic diagram (left) and the equivalent circuit of dc SQUID (right). I_c , which is the sum of the critical currents at the junctions, modulates periodically with a period of $\Phi_0 = h/2e$ once an external magnetic flux $H = B/\mu_0$ is



Fig. 2.14 The integral path on the superconducting ring of dc SQUID

applied to the superconducting ring. By reading I_c , we can precisely obtain the microscopic flux applied to the dc SQUID. More generally, the voltage across the ring is read by biasing the SQUID with a constant current above I_c . A large I_c allows a large super-current flow, which makes the voltage across the ring small, while a small I_c makes the voltage across the ring large. The dc SQUID, therefore, is a flux-to-voltage converter.

2.3.5 Principles of dc SQUID

Let us consider integral paths $1 \to 2$ and $1' \to 2'$ on the superconducting ring of dc SQUID (Figure 2.14). Let $\phi(1), \phi(2), \phi(1')$ and $\phi(2')$ be phases at the junctions. We compute integrals along the paths and get

$$\phi(1) - \phi(2) = \frac{2\pi}{\Phi_0} \mu_0 \lambda_{\rm L}^2 \int_2^1 \vec{j}_{\rm s} d\vec{l} + \frac{2\pi}{\Phi_0} \int_2^1 \vec{A} d\vec{l}$$
(2.124)

and

$$\phi(2') - \phi(1') = \frac{2\pi}{\Phi_0} \mu_0 \lambda_{\rm L}^2 \int_{1'}^{2'} \vec{j}_{\rm s} d\vec{l} + \frac{2\pi}{\Phi_0} \int_{1'}^{2'} \vec{A} d\vec{l}, \qquad (2.125)$$

where $\lambda_{\rm L}$ the London penetration depth, \vec{j}_s is the critical current density of superconductor, and \vec{A} is the vector potential of applied magnetic flux. Summing (2.124) and (2.132) and adding $(2\pi/\Phi_0) \left[\int_{2'}^2 \vec{A} d\vec{l} + \int_1^{1'} \vec{A} d\vec{l} \right]$ to the both sides we have

$$\phi(2') - \phi(2) - \frac{2\pi}{\Phi_0} \int_2^{2'} \vec{A} d\vec{l} - \left(\phi(1') - \phi(1) - \frac{2\pi}{\Phi_0} \int_1^{1'} \vec{A} d\vec{l}\right) \\ = \frac{2\pi}{\Phi_0} \oint \vec{A} d\vec{l} + \frac{2\pi}{\Phi_0} \mu_0 \lambda_{\rm L}^2 \left(\int_2^1 \vec{j}_{\rm s} d\vec{l} + \int_{1'}^{2'} \vec{j}_{\rm s} d\vec{l}\right). \quad (2.126)$$

We can ignore the integrals at the left side because of the size ratio of superconductors and insulators. Defining $\delta_1 \equiv \phi(1') - \phi(1)$ and $\delta_2 \equiv \phi(2') - \phi(2)$ yields

$$\delta_2 - \delta_1 = \frac{2\pi}{\Phi_0} \left\{ \Phi + \mu_0 \lambda_{\rm L}^2 \left(\int_2^1 \vec{j}_{\rm s} d\vec{l} + \int_{1'}^{2'} \vec{j}_{\rm s} d\vec{l} \right) \right\} = \frac{2\pi}{\Phi_0} \Phi_{\rm T}, \qquad (2.127)$$

where $\Phi_{\rm T}$ is the total flux through the superconducting ring. Let *L* be the self-inductance of the superconducting ring and *J* be the super-current in the ring, and $\Phi_{\rm T}$ then becomes

$$\Phi_{\rm T} = \Phi_{\rm a} + LJ, \tag{2.128}$$

where Φ_a is the externally applied magnetic flux.

Figure 2.13 (right) shows the equivalent circuit of dc SQUID using the RCSJ model. With a current at the junction J1 $I_1 = I/2 + J$ and a current at the junction J2 $I_2 = I/2 - J$, we have

$$\frac{I}{2} \pm J = I_{0,k} \sin \delta_k + \frac{\Phi_0}{2\pi R_k} \dot{\delta}_k + \frac{\Phi_0}{2\pi} C_k \ddot{\delta}_k + I_{N,k} \qquad \text{(for } k = 1,2\text{)}$$
(2.129)

for each junction. The voltages U_k (k = 1, 2) across the junctions are

$$U_k = \frac{\Phi_0}{2\pi} \dot{\delta}_k \qquad (k = 1, 2). \tag{2.130}$$

Here we define the averaged critical current $I_0 = (I_{0,1}+I_{0,2})/2$, (twice) the parallel resistance $R = 2R_1R_2/(R_1+R_2)$ and the capacitance $C = (C_1 + C_2)/2$, and normalize (2.129) and (2.127) using I_0 for currents, R for resistances, $\tau \equiv \Phi_0/(2\pi I_0 R) = \omega_c^{-1}$ for time, $I_0 R$ for voltages, and Φ_0 for magnetic flux. We then obtain

$$\frac{i}{2} + j = (1 - \alpha_I)\sin\delta_1 + (1 - \alpha_R)\dot{\delta}_1 + \beta_c(1 - \alpha_C)\ddot{\delta}_1 + i_{N,1}, \qquad (2.131)$$

$$\frac{i}{2} - j = (1 + \alpha_I)\sin\delta_2 + (1 + \alpha_R)\dot{\delta}_2 + \beta_c(1 + \alpha_C)\ddot{\delta}_2 + i_{N,2}, \qquad (2.132)$$

and

$$\delta_2 - \delta_1 = 2\pi (\phi_a - \frac{1}{2}\beta_L j), \qquad (2.133)$$

where $i \equiv I/I_0$, $j \equiv J/I_0$, $i_{N,k} \equiv I_{N,k}/I_0$ (k = 1, 2), and $\phi_a = \Phi_a/\Phi_0$. α_I , α_R , and α_C are parametrize asymmetries in the junction critical currents, resistances and capacitances, respectively. Dots above variables are derivatives with respect to τ . β_L is called the screening parameter defined as

$$\beta_{\rm L} \equiv \frac{2LI_0}{\Phi_0}.\tag{2.134}$$

The normalized voltages u_k are given by $u_k = \dot{\delta}_k$ (k = 1, 2).

We now consider a static solution when two junctions are identical as a simple limit case. (2.131) and (2.132) now become

$$\frac{i}{2} + j = \sin \delta_1, \qquad \frac{i}{2} - j = \sin \delta_2.$$
 (2.135)



Fig. 2.15 Critical current of the dc SQUID vs. applied magnetic flux for various values of $\beta_{\rm L}$ (left) and the modulation depth vs. $\beta_{\rm L}$ (solid line is $\beta_{\rm L}^{-1}$) (right). Adapted from Clarke and Braginski [6].



Fig. 2.16 Φ -V characteristic: $\beta_{\rm C} = 0$ (left) and $\beta_{\rm C} = 1$ (right). Adapted from [6].

Moreover, when the self-inductance of the SQUID is negligible, $\beta_{\rm L} \ll 1$, thus (2.133) becomes

$$\delta_2 - \delta_1 = 2\pi\phi_a. \tag{2.136}$$

We now have

$$i = \sin \delta_1 + \sin \delta_2 = \sin \delta_1 + \sin(\delta_1 + 2\pi\phi_a).$$
 (2.137)

By defining $\gamma \equiv \delta_1 + \pi \phi_a$, we finally have

$$i = 2\sin\gamma \cdot \cos(\pi\phi_{\rm a}). \tag{2.138}$$

The critical current, which is the maximum super-current, is therefore $i_c = 2 |\cos(\pi \phi_a)|$ when normalized, or simply

$$I_{\rm c} = 2I_0 \left| \cos \left(\pi \frac{\Phi_{\rm a}}{\Phi_0} \right) \right|. \tag{2.139}$$

In this limit case, the critical current modulates from 0 to $2I_0$.

Practically, the self-inductance L can not be ignored. L reduces the modulation depth $\Delta I_c/I_{c,\max}$ by $1/(\beta_L+1)$. When the magnetic flux applied externally is $\Phi_0/2$, the required ring current to quantize the flux in the ring is the order of $J = \Phi_0/2L$. Then, the minimum critical current is the order of $2(I_0 - J)$, and $\Delta I_c/I_{c,\max}$ is therefore

$$\Delta I_{\rm c}/I_{\rm c,max} \approx \frac{I_{\rm c,max} - 2(I_0 - J)}{I_{\rm c,max}} = \frac{\Phi_0}{2LI_0} = \frac{1}{\beta_{\rm L}},\tag{2.140}$$

meaning that the modulation depth is reduced by $1/\beta_L$. Figure 2.15 shows the critical current modulation depth for various values of β_L .

We now consider the case of nonzero dc voltage across the junctions. We first consider the case when $\beta_{\rm L} \ll 1$. Assuming two junctions are identical, (2.131) and (2.132) become

$$\frac{i}{2} + j = \sin \delta_1 + \dot{\delta}_1 + \beta_c \ddot{\delta}_1 \tag{2.141}$$

$$\frac{i}{2} - j = \sin \delta_2 + \dot{\delta}_2 + \beta_c \ddot{\delta}_2. \tag{2.142}$$

As $\beta_{\rm L} \ll 1$, we obtain $\dot{\delta}_1 = \dot{\delta}_2$ and

$$i = \sin \delta_1 + \sin(\delta_1 + 2\pi\phi_a) + 2\dot{\delta}_1 + 2\beta_c\ddot{\delta}_1$$
 (2.143)

from (2.133). For $\gamma = \delta_1 + \pi \phi_a$, we have $i = 2(\cos \pi \phi_a \cdot \sin \gamma + \dot{\gamma} + \beta_c \ddot{\gamma})$, or in general

$$I = 2I_0 \cos\left(\pi \frac{\Phi_a}{\Phi_0}\right) \sin\gamma + \frac{2\Phi_0}{2\pi R} \dot{\gamma} + \frac{2\Phi_0}{2\pi} C \ddot{\gamma}.$$
(2.144)

Figure 2.16 (left) shows the numerical simulation of the Φ -V characteristic at $\beta_{\rm C} = 0$ [6]. The above equation is equivalent to (2.110) if the resistance, the capacitance, and the critical current are substituted for R/2, 2C, and $2I_0 \cos \pi \phi_{\rm a}$, respectively. In the limit of $\beta_{\rm c} \ll 1$, the I-V characteristic of dc SQUID for $I > I_{\rm c}$ is therefore

$$V = \frac{R}{2}\sqrt{I^2 - I_{\rm c}^2}.$$
 (2.145)

This means that the voltage across dc SQUID modulates periodically with a period of Φ_0 as an external magnetic flux is applied to the SQUID. Partially differentiating the above with a respect to Φ_a , we have

$$\frac{\partial V}{\partial \Phi_{\rm a}} = -2\pi \frac{I_0 R}{\Phi_0} \frac{I_0 \sin\left(\pi \Phi_{\rm a}/\Phi_0\right) \cos\left(\pi \Phi_{\rm a}/\Phi_0\right)}{\sqrt{I^2 - I_{\rm c}^2}}.$$
(2.146)

It diverges at $I = I_c$, however, it converges when the noise is not negligible and the maximum will be around $I = I_c$. Let us now define the SQUID transfer function V_{Φ} as

$$V_{\Phi} = \max\left(\left|\frac{\partial V}{\partial \Phi_{\rm a}}\right|\right). \tag{2.147}$$

From (2.145) the peak-to-peak output voltage $V_{\rm pp}$ is

$$V_{\rm pp} = V(\Phi_{\rm a} = \Phi_0/2) - V(\Phi_{\rm a} = 0) = I_0 R \left[\frac{I}{2I_0} - \sqrt{\left(\frac{I}{2I_0}\right)^2 - 1} \right],$$
(2.148)

which takes its maximum value $I_0 R$ when $I = 2I_0$.

When $\beta_{\rm L}$ can not be ignored, the critical current is reduced by $1/(\beta_{\rm L}+1)$, thus $\partial V/\partial \Phi_{\rm in}$ is reduced by

$$\frac{\partial V}{\partial \Phi_{\rm a}} \propto \frac{1}{\beta_{\rm L} + 1},$$
(2.149)

and V_{Φ} is also reduced by

$$V_{\Phi} \propto \frac{1}{\beta_{\rm L} + 1}.\tag{2.150}$$

It does not mean that $\beta_L = 0$ is the optimized SQUID. We will discuss it later, but $\beta_L \approx 1$ will be the optimized

considering the effect of noise.

We finally see the effect of capacitance C to the SQUID transfer function. When the Josephson frequency $f_{\rm J} = \omega_{\rm c}/2\pi = V/\Phi_0$ becomes the same level of the resonance frequency $1/2\pi\sqrt{L(C/2)}$, the effect of L-C resonance can not be negligible. When $\Phi_{\rm a} = 0$, the phases at two junctions match and it does not start oscillating. However, if they are out-of-phase, it starts oscillating. Once the oscillation occurred, the output voltage becomes smaller as the dc component of Josephson current becomes larger. As it works opposite to the transfer function, the SQUID output becomes smaller when $\Phi_{\rm a} = \Phi_0/2$, which the output is supposed to be the maximum, while the output stays same when $\Phi_{\rm a} = 0$. The $\Phi-V$ characteristic then becomes severely distorted as shown in Figure 2.16 (right).

2.3.6 Johnson-Nyquist Noise

The power spectral density S_V of Johnson noise at the shunt resistors of junctions can be given by

$$S_V = 4k_{\rm B}TR,\tag{2.151}$$

and it causes fluctuations in the output voltage and the ring current. While there is no correlation between the output voltage and the ring current in the case of a non-superconducting ring $(I_0 = 0)$, in the SQUID there is a correlation between them since δJ , the fluctuation of the ring current J, causes $\delta \Phi$, the fluctuation of the magnetic flux, and the output voltage V is the function of the flux.

If we normalize S_V using $\omega_c = \tau^{-1} = 2\pi I_0 R / \Phi_0$ and $V_c = I_0 R$, we obtain

$$s_V = S_V \frac{\omega_c}{V_c^2} = 4\Gamma \tag{2.152}$$

with the noise parameter Γ .

Using the SQUID transfer function V_{Φ} , the magnetic flux noise power S_{Φ} is given by

$$S_{\Phi} = \frac{S_V}{V_{\Phi}^2},\tag{2.153}$$

which is also one of the important figures of merit of SQUID.

With (2.140), (2.150) and the effect of $\beta_{\rm C}$ to $\Phi - V$ characteristic, a symmetric dc SQUID is now characterized by three parameters, $\beta_{\rm C}$, $\beta_{\rm L}$ and Γ . In the next section, we discuss the optimization of these parameters.

2.3.7 Optimization

Here we review the definitions of the three parameters:

$$\beta_{\rm C} \equiv \frac{2\pi}{\Phi_0} I_0 R^2 C, \qquad \beta_{\rm L} \equiv \frac{2LI_0}{\Phi_0}, \qquad \Gamma \equiv \frac{2\pi k_{\rm B} T}{I_0 \Phi_0}. \tag{2.154}$$

It is obvious that the optimized $\beta_{\rm C}$ is $\beta_{\rm C} \ll 1$ considering undesirable effects of hysteretic junctions and the LC resonance at the superconducting ring. To suppress the thermal fluctuation, the noise parameter should be $\Gamma \ll 1$. A typical Γ at liquid helium temperature is

$$\Gamma = 0.018 \left(\frac{T}{4.2 \,\mathrm{K}}\right) \left(\frac{10 \,\mu\mathrm{A}}{I_0}\right),\tag{2.155}$$



Fig. 2.17 $\beta_{\rm L}$ optimization. Adapted from Tesche and Clarke [31].

which is already very small, so we may not need to worry about Γ at liquid helium temperature. At liquid nitrogen temperature, Γ becomes 18 time larger, and is generally not negligible.

As seen in (2.140) and (2.150), $\beta_{\rm L}$ reduces the output voltage, but $\beta_{\rm L} \ll 1$ is not preferable. Figure 2.17 shows the computed values of $S_{\Phi}/2L$, which is equivalent to the current noise, as the function of $\beta_{\rm L}$ [31]. At $\beta_{\rm L} \ll 1$, $S_{\Phi}/2L$ takes a large value, and then takes the minimum value around $\beta_{\rm L} \approx 1$, so the optimized value for $\beta_{\rm L}$ is $\beta_{\rm L} \approx 1$.

To summarize, the optimized values for $\beta_{\rm C}$, $\beta_{\rm L}$ and Γ are

$$\beta_{\rm C} \ll 1, \qquad \beta_{\rm L} \approx 1, \qquad \Gamma \ll 1.$$
 (2.156)

2.3.8 Numerical Analysis and Approximate Solutions for dc SQUID

Equations from (2.131) to (2.133) are numerically analyzed in order to discuss the SQUID performance.

The transfer function V_{Φ} , the voltage noise power spectral density S_V , and the magnetic flux noise power spectral density S_{Φ} at the operating point $I \approx 1.6I_0$ and $\Phi_a = 0.25\Phi_0$, at which the SQUID gain is maximum, for the optimized SQUID, $\beta_C \ll 1$, $\beta_L \approx 1$, and $\Gamma = 0.05$, are approximated as

$$V_{\Phi} \approx \frac{R}{L},\tag{2.157}$$

$$S_V \approx 16k_{\rm B}TR,\tag{2.158}$$

$$S_{\Phi} \approx \frac{16k_{\rm B}TL^2}{R} \tag{2.159}$$

[4, 5, 33, 27].

The above approximations do not hold on the SQUID at liquid nitrogen temperature. In this case the gener-



Fig. 2.18 TES readout using SQUID

alized approximations [10, 9] below are often used:

$$V_{\Phi} = \frac{4I_0R}{\Phi_0(1+\beta_{\rm L})} \exp\left(-\frac{3.5\pi^2(\delta\Phi_{\rm n})^2}{\Phi_0^2}\right) = \frac{4I_0R}{\Phi_0(1+\beta_{\rm L})} \exp(-2.75\Gamma\beta_{\rm L})$$
(2.160)

$$S_{\Phi} = \alpha L^2 \left(\frac{2k_{\rm B}T}{R}\right) \left[1 + \left(\frac{R}{LV_{\Phi}}\right)^2\right]$$
(2.161)

 $(\delta \Phi_n)^2 = k_B T L$ is the mean square of the magnetic flux noise, and $\alpha = 1 + \exp(1.23 - 4.82\Gamma)$.

The dynamic resistance R_{dyn} is approximated as

$$R_{\rm dyn} = \frac{R}{\sqrt{2}} \tag{2.162}$$

from the numerical analysis of the $\Phi - V$ characteristic [6].

2.3.9 TES Readout with SQUID

To read a current change of TES in the voltage-biasing mode, one needs an ammeter. Since the TES current is as small as ~ 100 μ A, and the TES impedance is as small as tens of milliohms, it needs to have a resolution of microamps, a low input impedance of milliohms. It should also have a low equivalent input noise, preferably less than 20 pA/ $\sqrt{\text{Hz}}$, which is the typical TES current noise. This makes no other choice other than the SQUID for the ammeter. Figure 2.18 shows the schematic diagram of TES readout circuit using the SQUID. The current from TES is converted to a magnetic flux through an input coil, which is around tens of pH to a few nH, and then is converted to a voltage that can be read by ordinary voltmeters. In this way the SQUID is used as a current to voltage converter, or a transimpedance amplifier.

Using the mutual-inductance $M_{\rm in}$ between the input coil and the SQUID superconducting ring, the current to voltage transfer function of SQUID, namely the transimpedance gain $Z_{\rm tran}$, is given by

$$Z_{\rm tran} = M_{\rm in} V_{\Phi}. \tag{2.163}$$



Fig. 2.19 The schematic diagram of Flux-Locked Loop (FLL)

Using the above, the equivalent input current noise of SQUID, $I_{\rm N}$, is then given by

$$I_{\rm N} = \frac{\sqrt{S_V}}{Z_{\rm tran}} = \frac{\sqrt{S_\Phi}}{M_{\rm in}}.$$
(2.164)

2.3.10 Flux-Locked Loop

Since the SQUID has a periodic response, the transfer function largely vary depend on the operating point. Moreover, the SQUID output turns around for a large input, which makes difficult to use as an ordinary amplifier. For that reason the SQUID is generally used with a negative feedback. The feedback keeps the flux through the SQUID ring constant, and is therefore called a flux-locked loop (FLL). In FLL, the SQUID output is fed back to a feedback coil, which is coupled magnetically with the SQUID superconducting ring, through a feedback resistor (Figure 2.19).

In this case the feedback factor b is given by

$$b = \frac{\Phi_{\rm FB}}{V_{\rm out}} = \frac{M_{\rm FB}}{R_{\rm FB}},\tag{2.165}$$

and the overall gain of the FLL circuit becomes $1/b = R_{\rm FB}/M_{\rm FB}$, where $R_{\rm FB}$ is the feedback resistance and $M_{\rm FB}$ is the mutual-inductance between the feedback coil and the superconducting ring. The magnetic flux $\Phi_{\rm in}$ that the input current $I_{\rm in}$ generates is

$$\Phi_{\rm in} = M_{\rm in} I_{\rm in} \tag{2.166}$$

using $M_{\rm in}$, the mutual-inductance between the input coil and the superconducting ring. The (current-to-voltage) transfer function Ξ in the FLL circuit is then given by

$$\Xi = \frac{M_{\rm in}}{M_{\rm FB}} R_{\rm FB}.$$
(2.167)

2.3.11 SQUID Array

By joining multiple SQUID in series and having them work in phase, one can have an increased SQUID output. It is called a SQUID series array. The series SQUID array tends to have a large output impedance, so one may



Fig. 2.20 The TES readout using the SQUID array: the two-staged SQUID (left), and the SQUID series array (right)

have to parallelize the series SQUID array to reduce the impedance.

Let us consider a SQUID array with n series and m parallel. With the single SQUID transfer function V_{Φ} , the voltage noise power spectral density S_V , the magnetic flux noise power spectral density S_{Φ} , and the dynamic resistance R_{dyn} , we have the SQUID array transfer function \mathcal{V}_{Φ} , the voltage noise spectral density \mathcal{S}_V , the magnetic flux noise power spectral density \mathcal{S}_{Φ} , and the dynamic resistance \mathcal{R}_{dyn} as

$$\mathcal{V}_{\Phi} = nV_{\Phi},\tag{2.168}$$

$$S_V = -\frac{n}{m} S_V, \qquad (2.169)$$

$$\mathcal{S}_{\Phi} = \frac{\mathcal{S}_V}{\mathcal{V}_{\Phi}^2} = \frac{S_{\Phi}}{nm},\tag{2.170}$$

and

$$\mathcal{R}_{\rm dyn} = \frac{n}{m} R_{\rm dyn}, \tag{2.171}$$

respectively. From $Z_{\text{tran}} = M_{\text{in}}V_{\Phi}$, we thus have the SQUID array transimpedance gain $\mathcal{Z}_{\text{tran}}$ as

$$\mathcal{Z}_{\text{tran}} = M_{\text{in}} \mathcal{V}_{\Phi} = n M_{\text{in}} V_{\Phi}. \tag{2.172}$$

The SQUID output becomes n time larger in the series array. The voltage noise also becomes n time larger in the series array, but m time smaller in the parallel array. The equivalent input current noise \mathcal{I}_N in the array now becomes

$$\mathcal{I}_{\mathrm{N}} = \frac{\sqrt{\mathcal{S}_{V}}}{\mathcal{Z}_{\mathrm{tran}}} = \sqrt{\frac{1}{nm}} I_{\mathrm{N}}, \qquad (2.173)$$

which is reduced by $\sqrt{1/nm}$. Therefore, as the number of SQUID increases either in series or in parallel, the overall noise level decreases and the signal-to-noise ratio increases. This is a huge advantage of the SQUID array. The number of SQUID in array can be as many as tens or hundreds.

There are generally two types of amplifiers using the SQUID array; one is a two-staged SQUID amplifier shown in Figure 2.20 (left), and the other is a single-stage SQUID series amplifier shown in Figure 2.20 (right). The former is called the TSS (two-staged SQUID) amplifier, and the later is called the SSA (series SQUID array) amplifier.

2.3.12 Joule Heating

The SQUID at the operating point generates heat. The heat is Joule heating at the shunt resistors of junctions, and the maximum heat P is

$$P = VI = 2RI_0^2 \tag{2.174}$$

at $I = 2I_0$ and $V = I_0 R$, at which the SQUID output becomes maximum. The Joule heating for the SQUID array \mathcal{P} with *n* series and *m* parallel is the sum of Joule heating of each SQUID as

$$\mathcal{P} = nmP. \tag{2.175}$$

2.3.13 Bandwidth

Because the SQUID voltage output is the periodic function with a period of Φ_0 , a flux input exceeding roughly $\pm 1/4\Phi_0$ causes a Φ_0 jump of the operating point, which is called a flux jump. The maximum flux input Φ_{max} into SQUID without flux-jumping is given by

$$\frac{1}{1+\mathcal{L}(\omega)} \left| \Phi_{\max} \right| < \frac{1}{4} \Phi_0, \tag{2.176}$$

where $\mathcal{L}(\omega)$ is the loopgain of FLL.

2.3.14 SQUID Noise Contribution to Energy Resolution

The NEP of the SQUID is given by

$$NEP_{readout}^2 = \frac{i_n^2}{S_I^2},$$
(2.177)

where i_n is the noise current spectral density of SQUID. The contribution of the SQUID noise to the energy resolution is thus

$$\Delta E_{\rm FWHM} = 2.35 \left(\int_0^\infty \frac{4df}{\text{NEP}_{\rm readout}^2(f)} \right)^{-\frac{1}{2}}$$
(2.178)

$$= 2.35 \frac{\mathcal{L}_0 + 1}{\mathcal{L}_0} |b| i_n \sqrt{\tau_{\text{eff}}}$$
(2.179)

$$= 2.35 \frac{\mathcal{L}_0 + 1}{\mathcal{L}_0} V_{\rm b} i_n \sqrt{\tau_{\rm eff}}$$
(2.180)

using (2.88). Therefore, for the case of $\mathcal{L}_0 \gg 1$, it is

$$\Delta E_{\rm FWHM} \sim 2.35 V_{\rm b} i_n \sqrt{\tau_{\rm eff}}.$$
(2.181)



Fig. 2.21 The schematic diagram of the SQUID TDM architecture. Adapted from Irwin et al. [18].

2.3.15 SQUID Multiplexing

To readout a TES using a single SQUID, we generally need 4 pairs of electrical wires; one pair for TES biasing, one for a SQUID feedback, one for SQUID biasing, and one for a SQUID output. As the number of TES increases, the required number of SQUID increases, meaning the required number of electrical wires also increases, which finally causes non-negligible heat flow from a higher temperature stage to a cryogenic stage. Therefore, the maximum number of SQUID that can be placed at a cryogenic stage is generally limited by a cooling power for the stage, and is typically less than 100 even by optimistic estimates. For readout of more than hundreds of TES, SQUID multiplexing is necessary.

The methods of SQUID multiplexing are divided into several types: time-division multiplexing (TDM), codedivision multiplexing (CDM), and frequency-division multiplexing (FDM).

Time-Division Multiplexing

In TDM, each TES signal is connected to each SQUID, and SQUIDs are turned on one at a time. For example, Figure 2.21 shows the schematic diagram of TDM developed by Irwin et al. [18]. Each TES is coupled to each SQUID, which is shunted with an address resistor and a coil coupled to a summing coil that connects all of the SQUIDs in a column. The summing coil is then coupled to a single second-stage SQUID, and the output of the second-stage SQUID is finally coupled to a third-stage SQUID array at a higher temperature. Address currents, $I_M(t)$, are turned on one row at a time. By switching address currents far faster than a time constant of TES, one can readout multiple TES signals with a single (column of) SQUID.

Although one can theoretically multiplex any number of signals with TDM as long as SQUIDs are switched fast enough, it also increases the noise because of SQUID noise aliasing [16]. If the number of multiplexing is N, then



Fig. 2.22 Signal summing methods in FDM: flux summing (left), voltage summing (middle), and current summing (right)

the effective noise power of SQUID is increased by a factor of N. In order to keep the SNR of a non-multiplexed case, the SQUID gain needs to be N times larger, making it difficult to increase the number of multiplexing.

Code-Division Multiplexing

CDM is an enhanced version of TDM. In CDM, SQUIDs are not turned on/off but switched polarity, or modulated, according to the Walsh code. In contrast to TDM, all the TES are biased all the time in CDM. As a result, both the effective signal power and the effective noise power are increased by a factor of N, and the SNR is thus maintained. Moreover, all signals except the unmodulated channel are moved to higher signal bands of modulation frequencies, and the demodulated signals are free of low-frequency noises, such as 1/f noise and power line noise. Therefore, the modulated channels usually exhibit better energy resolutions than the unmodulated channel.

As the cryogenic system of CDM is identical to that of TDM, CDM is used as a better replacement of TDM. Therefore, CDM is adopted by groups that developed TDM. For example, NIST has achieved 2.6 eV FWHM at 5.9 keV multiplexing 8 channels [11].

Frequency-Division Multiplexing

In FDM, TES are AC-biased in different frequencies, and their signals are summed, or modulated, at a single SQUID (or SQUID array). The modulated signals are demodulated at a room-temperature electronics to retrieve TES signals of each channel. Biasing frequencies need to be faster than the thermal diffusion time constant and are generally several MHz. In this method a single SQUID is only used, and there is no noise aliasing nor an addition of multiple SQUID noises. Consequently, the SNR is maintained as in a non-multiplexed case. Moreover, all signals are moved to higher signal bands, as in CDM, and demodulated signals are also free of low-frequency noises. In FDM, the required number of SQUID is less than that in TDM and CDM, and the cryogenic stage is therefore rather simpler, although band-pass filters with different passbands are required for each TES. One disadvantage in FDM is that a large slew rate and a large dynamic range are required to SQUID since the input to SQUID is larger than TES signals as TES are AC-biased with amplitudes evidently more than amplitudes of TES signals.

FDM can be further divided into three types by its signal summing methods: flux summing, voltage summing (or summing loop), and current summing. In flux summing, signals are converted to flux then summed at a SQUID input (Figure 2.22 left). Owing to spatial limitations of multiple input coils and a SQUID washer, the number of multiplexing is limited. In voltage summing, signals are summed using a superconducting inductance loop called a summing loop (Figure 2.22 middle). It has been developed by a group of UCB and LLNL [7]. In current summing, signals are summed directly at a SQUID input (Figure 2.22 right). It is the simplest method, but has a disadvantage that a signal from a TES may goes to other signal lines, which causes signal cross talks, due to a non-zero common impedance at a summing point. However, the input impedance is mostly nulled with FLL, and cross talks due to the common impedance is generally negligible. This method is widely adopted including TES bolometer FDM readouts [e.g. 14, 13, 8] as well as TES X-ray FDM readouts [e.g. 32]. We developed the 8-input flux-summing SQUID before, but we now have also moved to current-summing SQUIDs.

Comparison of Multiplexing Types

CDM and FDM share most of benefits, and can be good candidates for practical TES multiplexing applications. However, there is one non-negligible disadvantage in CDM especially for space missions. In CDM, TES bias lines can not be multiplexed, and the required number of wires in CDM is larger than that in FDM, which makes the required cooling power of cryogenic stage larger in CDM. It is critical disadvantage for space missions where resources are very limited. Therefore, FDM is generally the only choice to multiplex large-format TES arrays in space missions.

Chapter 3

Development of Low-Power SQUIDs

3.1 Requirements for SQUIDs

DIOS is planned to carry a 256 to 400 channel TES array to carry out the WHIM mapping. To readout the 256 channel TES array multiplexed by, for instance, 8, we need at least 32 SQUIDs. The maximum heat allowance at the cryogenic stage of the DIOS ADR is 640 nW, which makes the maximum heat dissipation of a single SQUID to be 20 nW or less. For two-staged configuration, the SQUID at the first stage is usually a single non-arrayed SQUID (or very small number of SQUID arrays) and this requirement can easily fulfilled. However, for single-staged configuration, the single SQUID needs to amplify the signal from TES usually up to 100 times or more, causing a lot of heat dissipation, which makes it very challenging to place it at the resource-limited cryogenic stage. Typical noise levels of TES and low-noise amp at room temperature are $10 \text{ pA}/\sqrt{\text{Hz}}$ and $1 \text{ nV}/\sqrt{\text{Hz}}$ respectively, which make an absolute minimum SQUID gain of 100 V/A (or 100Ω) or more, and the equivalent input current noise to be less than $10 \text{ pA}/\sqrt{\text{Hz}}$. To realize a single-staged SQUID readout, one needs an unprecedented low-power SQUID with an adequate gain.

Besides, to multiplex TES signals, which typical bandwidth is ≤ 100 kHz, in frequency domain, the TES and SQUID are AC-biased to up to several MHz, so they need to be optimized for operations in the MHz band. The wires between the cryogenic stage and room temperature electronics typically has some parasitic capacitance of $\sim 100 \text{ pF}$ or more, which attenuates the SQUID output and decreases the effective SQUID gain. Assuming the parasitic capacitance of 100 pF and the cut off frequency of 6 MHz^{*1} , the output resistance becomes 250Ω . The wires also have some resistance of $\sim 100 \Omega$. Therefore, the output impedance of SQUID is preferred to be 150Ω or less.

These design goals are summarized in Table 3.1 along with the minimum requirements if those are unfeasible.

		Goal	Requirement
Thermal power dissipation (nW)	P	< 20	< 32
Equivalent input current noise $@4 \text{ K} (pA/\sqrt{\text{Hz}})$	$I_{\rm N}$	< 10	< 20
Transimpedance gain $(V/A \text{ or } \Omega)$	$Z_{\rm tran}$	> 100	> 50
Output impedance (Ω)	Z_{out}	< 150	< 300

Table 3.1 Design targets of low-power SQUIDs

^{*1} It is equivalent to the required bandwidth for 16-channel multiplexing described in Section 4.2.

		ISAS-A10/C10	ISAS-B10	ISAS-G15
Junction critical current (μA)	I_0	10	10	8
Junction shunt resistance (Ω)	R	10	15	10
SQUID washer self-inductance (pH)	L	100	100	120
Input coil self-inductance (pH)	L_{in}	100	100	120
Input coil mutual-inductance (pH)	$M_{\rm in}$	75	75	90
Number of SQUIDs in series	n	10	10	15

Table 3.2 Design parameters of ISAS-A10, ISAS-B10, ISAS-C10, and new ISAS-G15

Table 3.3 Design targets and actual results of ISAS-A10, ISAS-B10 and ISAS-C10

		Goal	Requirement	Acti	ual Res	sults
				A10	B10	C10
Thermal power dissipation (nW)	P	< 20	< 32	21	33	26
Equivalent input current noise @ 4K (pA/\sqrt{Hz})	$I_{\rm N}$	< 10	< 20	8	10	8
Transimpedance gain (V/A or Ω)	$Z_{\rm tran}$	> 100	> 50	87	115	87
Output impedance (Ω)	$Z_{\rm out}$	< 150	< 300	96	141	103

3.2 Parameter Optimizations and SQUID Designs

3.2.1 Parameter Optimizations

As the gain of the SQUID becomes larger, the SQUID becomes hotter, meaning that a development of a cooler yet higher-gain SQUID is challenging. To build such low power SQUIDs with sufficient gains, we took following approach. The heat dissipation P and the transimpedance gain Z_{tran} are given by

$$P = 2nRI_0^2$$
 and $Z_{\rm tran} \propto nRI_0 M_{\rm in},$ (3.1)

where R is a junction shunt resistance, I_0 is the junction critical current, n is the number of SQUIDs in series, and $M_{\rm in}$ is a mutual-inductance of the SQUID washer and the input coil. Using these, the transimpedance gain per unit power, $Z_{\rm tran}/P$, is given by

$$Z_{\rm tran}/P \propto M_{\rm in}/I_0. \tag{3.2}$$

To maximize the gain per unit power, we can increase the mutual inductance, or decrease the critical current. However, the former limits the input dynamic range, and the latter makes the noise larger. As we place SQUIDs at the cryogenic stage which temperature is $\sim 100 \,\mathrm{mK}$, the SQUID equivalent input current noise, which is proportional to the temperature, tends to be smaller. Therefore, the latter approach is more feasible.

We have so far developed several types of low-power SQUIDs by decreasing the junction critical current, but none of them fully satisfied the requirement [28]. Table 3.2 shows the design parameters for low-power SQUIDs, ISAS-A10, ISAS-B10, and ISAS-C10. The junction critical current is $10 \,\mu$ A for all types. They also share most of the design parameters. The difference is the shunt resistance: ISAS-A10 and ISAS-C10 have $10 \,\Omega$ resistance, while ISAS-B10 has $15 \,\Omega$ resistance. ISAS-C10 is gradiometer type of ISAS-A10. Table 3.3 shows the measurement results of those SQUIDs. ISAS-A10 is sufficiently low-power, but has an insufficient gain. ISAS-B10 has a sufficient gain, but is slightly hotter.

To develop a low-power SQUID with a sufficient gain, we further decreased the critical current to $8 \,\mu$ A. Table 3.2 also shows the design parameters of a new ISAS-G15, which is a series array of 15 ISAS-G. Compared to ISAS-A10/C10, ISAS-G15 has slightly larger SQUID washer and input coil self-inductances, making the mutual-



Fig. 3.1 The design of the junction and shunt resistor part

inductance also slightly larger. As the heat dissipation of single SQUID is decreased due to the smaller critical current, the number of SQUIDs in series is increased to 15, but the heat dissipation is to be still under 20 nW. For other parameters, we have followed the SQUID optimization rule (2.156).

3.2.2 SQUID Designs

For the layout design we used Xic (Version 3.2) developed by Whiteley Research Inc. SQUIDs were fabricated in the Clean Room for Analog & Digital superconductiVITY (CRAVITY) at National Institute of Advanced Industrial Science and Technology (AIST), in accordance with the SRL Nb Standard Process (STD3). See Appendix A for the detail of the design parameter and rule in STD3.

Figure 3.1 shows the junction and shunt resistor part of ISAS-G15. The critical current density, 1 kA/cm^2 , and the junction design size including the shrink of junctions, $1.1 \,\mu\text{m} \times 1.1 \,\mu\text{m}$, make the critical current to be $8.1 \,\mu\text{A}$. The shunt resistor is bypassing the junction with $10 \,\Omega$ resistance.

Figure 3.2 shows the design of ISAS-G, a Ketchen-type [19] SQUID with a single-turn input coil and a single-turn feedback coil, and Figure 3.3 shows the design of ISAS-G15.

Finally, Figure 3.4 shows the design of a $2.5 \text{ mm} \times 2.5 \text{ mm}$ ISAS-G15 chip. It is equipped with two sets of ISAS-G15 with $5 \text{ m}\Omega$ TES shunt resistors, and a $15 \text{ k}\Omega$ SQUID heater for a removal of trapped flux. The TES shunt resistor is an interdigitated electrode array in shape to produce a very small resistance from the 2.4Ω sheet resistance.

3.2.3 Multi-Input Current-Summing SQUID

We also designed a 4-input current-summing SQUID for FDM. Figure 3.5 (left) shows the design of a 4-input ISAS-G15 chip with inductors. It is equipped with a ISAS-G15, a $5 \text{ m}\Omega$ TES shunt resistor, a $15 \text{ k}\Omega$ SQUID heater, and four 500 nH inductors. Figure 3.5 (right) shows the design of a 4-input extension chip with inductors. It is



Fig. 3.2 The design of ISAS-G

Table 3.4 Inductances, capacitances and resonance frequencies of the LC filters on the 4-input currentsumming SQUID and the 4-input extension chips

Inductance (nH)	Area (mm^2)	Capacitance (nF)	Resonance Frequency (MHz)				
4-input current-summing SQUID							
500	0.358	3.2	4.0				
	0.316	2.8	4.3				
	0.283	2.5	4.5				
	0.253	2.2	4.8				
4-input extension							
500	0.636	5.6	3.0				
	0.528	4.7	3.3				
	0.468	4.1	3.5				
	0.408	3.6	3.8				

equipped with four 500 nH inductors, which can be connected to the former SQUID chip to expand multiplexing channel up to 8. Figure 3.6 shows the equivalent circuit for those chips. Each TES channel is separated each other by a LC bandpass filter serially inserted to TES, and signals from each TES are summed right before the SQUID input coil. The capacitors for the filters are attached externally.

Externally attached capacitors for the LC filter are usually SMD multilayer ceramic capacitors. To design MHzband bandpass filters using on-chip 500 nH inductors, we need several nF capacitors. The typical size of those capacitors is usually 1608 (0603) or larger, which is almost the same size as the SQUID chip. To save precious cryogenic stage space, those capacitors should be also on chip. Thus, we designed a 4-input current-summing SQUID with built-in LC filters for FDM. Figure 3.7 shows the design of the chip along with an extension chip with built-in LC filters, and Figure 3.8 shows the equivalent circuit for those chips. The capacitors are simple parallel-plate capacitors. The dielectric is the JP (JJ protection) layer, which is a 10 nm thick anodic aluminum oxide film, and the relative permittivity is 8–10. Therefore, a 0.1 mm^2 sized capacitor will have $\sim 1 \text{ nF}$ in capacity. We summarized capacitances and resonance frequencies of the filters in Table 3.4.



Fig. 3.3 The design of ISAS-G15 $\,$



Fig. 3.4 The design of the ISAS-G15 chip

3.3 Measurements

All the SQUIDs were fabricated in the CRAVITY at AIST. Figure 3.9 shows photomicrographs of the fabricated ISAS-G15 chip. Figure 3.10 shows the photomicrographs of the 4-input current-summing SQUID chip with inductors, a zoomed 500 nH inductor, and a zoomed current-summing point. All the TES signal lines are paired on the chip, and only broken at the summing-point to sum those signals and put into the input coil. Figure 3.11 shows the photomicrographs of the fabricated 4-input current-summing SQUID chip with built-in LC filters, and its current-summing point. The TES signals are summed in the same manner. Finally Figure 3.12 shows the photomicrographs of the 4-input extension chip with inductors, and the 4-input extension chip with built-in filters.

The Φ -V characteristic was first measured for the fabricated ISAS-G15 in 4 K. From the measurement, we can derive most of the figures of merit except the noise. The noise characteristic and the filter characteristics were then measured also in 4 K.



Fig. 3.5 The designs of the 4-input ISAS-G15 chip with inductors (left) and the 4-input extension chip with inductors (right)



Fig. 3.6 Equivalent circuits for the 4-input SQUID chip with inductors and the 4-input extension chip with inductors



Fig. 3.7 The designs of the 4-input ISAS-G15 chip with built-in LC filters and the 4-input extension chip



Fig. 3.8 Equivalent circuits for the 4-input SQUID chip with built-in LC filters and the 4-input extension chip



Fig. 3.9 Photomicrographs of the fabricated ISAS-G15: an overall view of the chip (left) and a zoomed SQUID (right)



Fig. 3.10 Photomicrographs of the 4-input ISAS-G15 with inductors: an overall view of the chip (top), a zoomed inductor (bottom left), and a zoomed current-summing point (bottom right)



Fig. 3.11 Photomicrographs of the 4-input ISAS-G15 with inductors and capacitors: an overall view of the chip (left) and a zoomed current-summing point (right)



Fig. 3.12 Photomicrographs of the 4-input extension chips: an extension with inductors (left) and an extension with inductors and capacitors (right)

3.3.1 Setups

SQUIDs were measured using the Magnicon XXF-1. Figure 3.13 shows the experimental setups for Φ -V measurements and noise measurements. For both measurements, SQUIDs were placed at a cryogenic stage (Figure 3.15 left) of a 4 K probe (Figure 3.14) and shielded with a Cryoperm magnetic shield (Figure 3.15 right). The XXF-1 amp is placed in the head of the probe, which is a diecast aluminum box, and carefully shielded for a better noise performance. The SQUID and the amp are wired using paired constantan loom wires. The 4 K probe was then inserted into a LHe dewar upon measurements.

In $\Phi-V$ measurements, signals were captured using a Yokogawa DL708E DSO equipped with 701855 (12-bit ADC) modules. An IWATSU DS-5324 DSO was also used for a quick look of signals. The XXF-1 was powered by a TEXIO PW18-1.3AT stabilized power supply. In noise measurements, a HP 35670A FFT analyzer was used. In both measurements, all the powers were supplied from a DENKENSEIKI NCT-I noise cut transformer except for the XXF-1 controller PC. It is battery-powered on noise measurements, and optically-isolated from the system on $\Phi-V$ measurements.

3.3 Measurements



Fig. 3.13 Experimental setups for Φ -V measurements (left) and noise measurements (right)



Fig. 3.14 4 K probe



Fig. 3.15 The stage of the 4 K probe: the stage (left) and the stage covered with the Cryoperm magnetic shielding

In filter measurements, the developed analog front-end (Chapter 4) was used as the TES bias current supply and the low noise amp for the SQUID output.

3.3.2 $\Phi - V$ Measurements

An 1-Hz 60- μ A_{p-p} sine wave was applied to the SQUID through the feedback coil (or the input coil when measuring the mutual-inductance of the input coil), and SQUID output was captured 40 times and then averaged. The gain of the XXF-1 amp was set to 2000, and the 5 kHz LPF was applied when capturing at the DL708E. Figure 3.16 shows the obtained Φ -V characteristics of ISAS-G15 for bias currents from 8 μ A to 19 μ A. As the



Fig. 3.16 The Φ -V characteristics of ISAS-G15 for various values of bias current



Fig. 3.17 $\partial V/\partial \Phi_{in}$ and the dynamic resistance in 0–0.5 Φ_0 of ISAS-G15 for various values of bias current. The dashed lines show the design goals, while the solid lines show the design requirements.

minimum value of the Φ -V characteristic becomes larger than 0 when $I_{\rm b} > 17 \,\mu\text{A}$, the critical current of ISAS-G15 is calculated to be $8.5 \,\mu\text{A}$, which is slightly larger than the designed value. The power dissipation is then calculated to be slightly-hotter 22 nW. The mutual-inductances of the input coil and the feedback coil were calculated from periods of the Φ -V characteristic to be 83.26 pH and 76.62 pH for the input coil and the feedback coil, respectively.

From the $\Phi-V$ characteristics, the transfer function $\partial V/\partial \Phi_{in}$ and the dynamic resistance $\partial V/\partial I_b$ were also derived. $\partial V/\partial \Phi_{in}$ was obtained by chi-square fitting to $\Phi-V$ at each point from $0\Phi_0$ to $0.5\Phi_0$ for each bias current. Figure 3.17 (left) shows the obtained $\partial V/\partial \Phi_{in}$. The dashed line in the plot shows the design goal, $2.48 \text{ mV}/\Phi_0$, which is equivalent to the required transimpedance gain 100 V/A, and the solid line shows the design requirement, $1.24 \text{ mV}/\Phi_0$, which is equivalent to 50 V/A. The dynamic resistance was obtained by calculating $\partial V/\partial I_b$ from the obtained $\Phi-V$ characteristic. Figure 3.17 (right) shows the calculated dynamic resistance. The dashed line in the plot shows the design goal, 150Ω , and the solid line shows the design requirement, 300Ω .



Fig. 3.18 Equivalent input current noises of ISAS-G15 in the open loop for various Φ offsets: $I_{\rm b} = 16 \,\mu \text{A}$ (left) and $I_{\rm b} = 17 \,\mu \text{A}$ (right)

Bias current				Φ offset			
	$0.10\Phi_{0}$	$0.15\Phi_{0}$	$0.20\Phi_{0}$	$0.25\Phi_0$	$0.30\Phi_0$	$0.35\Phi_0$	$0.40\Phi_{0}$
	Outpu	ıt voltage	noise † $@$	4 K (nV	$V/\sqrt{\text{Hz}}$		
$15\mu\mathrm{A}$	1.0	1.0	1.6	1.5	1.5	1.4	1.3
$16\mu\mathrm{A}$	1.0	1.6	1.7	1.8	1.4	1.2	1.2
$17\mu\mathrm{A}$	2.0	2.8	2.0	1.4	1.2	1.1	1.2
	Equivalent	input cu	rrent nois	e [‡] @ 4 K	(pA/\sqrt{I})	Hz)	
$15\mu\mathrm{A}$		• • •	58	13	14	20	32
$16\mu\mathrm{A}$		57	13	17	20	27	41
$17\mu\mathrm{A}$	49	16	22	21	26	33	51
Equivalent inp	ut current	noise afte	er other no	oises subt	racted [‡] @	4 K (p.	A/\sqrt{Hz})
$15\mu\mathrm{A}$			42	9	10	12	17
$16\mu\mathrm{A}$		42	10	13	13	13	16
$17\mu\mathrm{A}$	42	15	18	13	13	10	20
4E	1 - V / TL						

	Table 3.5	Noise	characteristics	of	ISAS-G15	$^{\rm at}$	4 I
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 \dagger Errors are $< 0.1 \,\mathrm{nV}/\sqrt{\mathrm{Hz}}$

 \ddagger Errors are $< 1 \, \text{pA}/\sqrt{\text{Hz}}$

3.3.3 Noise Measurements

The SQUID output voltage noise in open-loop was measured at working points from $0.1\Phi_0$ to $0.4\Phi_0$ in $0.5\Phi_0$ steps for bias currents from $15 \,\mu\text{A}$ to $17 \,\mu\text{A}$. The gain of the XXF-1 amp was set to 2000, and the HP 35670A mode was set to FFT analysis with a uniform window in V/ $\sqrt{\text{Hz}}$ unit. The obtained noise was averaged 10 times in the 1–400 Hz band, 30 times in the 0.4–1.2 kHz band, and 50 times in the 1.2–100 kHz band.

Figure 3.18 shows the noise for bias currents $16 \,\mu\text{A}$ and $17 \,\mu\text{A}$. They are converted to the equivalent input current noise I_{N} by

$$I_{\rm N} = \frac{V_{\rm N}}{Z_{\rm tran}} = \frac{V}{GZ_{\rm tran}},\tag{3.3}$$

where $V_{\rm N}$ is the SQUID output voltage noise, $Z_{\rm tran}$ is the transimpedance gain calculated from the Φ -V characteristics, V is the system output voltage noise, and G is the XXF-1 amp gain. However, working points with $Z_{\rm tran} > 10 \,{\rm V/A}$ were only converted. SQUIDs usually have two types of noises, 1/f noise and white noise. Since we use our SQUIDs in the MHz band, we examine white noise only. Table 3.5 shows the white noise level at each



Fig. 3.19 The measured LC bandpass filter characteristic at 4 K

Table 3.6 The measured resonance frequencies of the LC filters on the 4-input current-summing SQUID and the 4-input extension chips

			Resonance Frequency			
Inductance (nH)	Area (mm^2)	Capacitance (nF)	Design (MHz)	Actual (MHz)		
	4-input	t current-summing S	SQUID			
500	0.358	3.2	4.0	5.294		
	0.316	2.8	4.3	5.629		
	0.283	2.5	4.5	5.959		
	0.253	2.2	4.8	6.291		
		4-input extension				
500	0.636	5.6	3.0	3.928		
	0.528	4.7	3.3	4.309		
	0.468	4.1	3.5	4.592		
	0.408	3.6	3.8	4.914		

working point obtained by fitting the noise spectra for f > 10 kHz, along with the calculated equivalent input current noise.

The noise values in Table 3.5 includes noises arisen from components other than SQUID. To eliminate those noises and estimate unblended SQUID noises, we first measured the system noise without having any SQUID at the cryogenic stage but shorted, and subtracted it from the measured noises. The system noise without SQUID is $2.2 \,\mu\text{V}/\sqrt{\text{Hz}}$. Table 3.5 also shows the subtracted noises.

3.3.4 Filter Measurements

The 4-input SQUID with built-in filters and the 4-input extension with built-in filters are connected by bonding wires, and eight $50 \text{ m}\Omega$ resistors are connected to those chips as a dummy TES. A sinusoidal signal was then applied to the TES bias line. Since the SQUID was used in the open-loop, the amplitude of the signal is kept low so that the SQUID output at the resonance frequencies is not folded. The frequency of the signal was then swept from 3 to 7 MHz in 1 kHz steps. Figure 3.19 shows the measured filter characteristic at 4 K. The designed resonance frequencies are from 3 to 5 MHz, but the measured resonance frequencies are from about 4 to 6.5 MHz, meaning that the actual relative permittivity is somewhat smaller than the expected value. Table 3.6 shows the measured resonance frequencies along with the designed values. The calculated actual relative permittivity is

				01	
$I_{\rm b}$	Φ offset	$\partial V / \partial \Phi_{ m in}$	$Z_{ m tran}$	$R_{\rm dyn}$	$I_{ m N}$
(μA)	(Φ_0)	(mV/Φ_0)	$(V/A \text{ or } \Omega)$	(Ω)	(pA/\sqrt{Hz})
15	0.30	2.9 ± 0.1	117 ± 1	152 ± 4	9.7 ± 0.2

Table 3.7 ISAS-G15 working point

Table 3.8Design targets and measurement results of ISAS-G15

		Measurement Result	Goal	Requirement
SQUID Parameter				
Critical current (μA)	I_0	8.5	_	_
Input coil mutual-inductance (pH)	$M_{\rm in}$	83.26	_	_
Feedback coil mutual-inductance (pH)	$M_{\rm FB}$	76.62	_	_
SQUID Figures of Merit				
Thermal power dissipation (nW)	P	22	< 20	< 32
Equivalent input current noise @ 4K (pA/\sqrt{Hz})	$I_{\rm N}$	10	< 10	< 20
Transimpedance gain (V/A or Ω)	$Z_{\rm tran}$	117	> 100	> 50
Output impedance (Ω)	Z_{out}	150	< 100	< 200

 $\sim 6.$

3.3.5 Working Points

From the $\Phi-V$ characteristic and the noise characteristics, the SQUID working points that fulfill the requirement were selected. There is only one working point found that fulfills all the requirements. Table 3.7 shows the working point and figures of merit at the point.

3.3.6 Summary

Table 3.8 summarizes the measurement results of ISAS-G15 along with the design targets.

Chapter 4

Development of Digital Electronics for Frequency-Division Multiplexing

4.1 Principles of Frequency-Division Multiplexing and Baseband Feedback

4.1.1 Preparation

We use the following definition for the Fourier transform:

$$\hat{f}(\omega) = \int_{-\infty}^{\infty} f(t) \mathrm{e}^{-i\omega t} dt = \mathcal{F}[f(t)].$$
(4.1)

The Fourier transform of $f(t)\cos(at)$ is then given by

$$\mathcal{F}[f(t)\cos(at)] = \frac{1}{2}\mathcal{F}[f(t)(e^{iat} + e^{-iat})] = \frac{1}{2}\hat{f}(\omega - a) + \frac{1}{2}\hat{f}(\omega + a).$$
(4.2)

We also define the resistance of a microcalorimeter at time t as

$$R(t) = R_0 + \Delta R(t), \tag{4.3}$$

where R_0 is the resistance at the operating temperature, and $\Delta R(t)$ is the deviation of the resistance due to an incident X-ray photon at time t.

4.1.2 Principles of AC-biased Microcalorimeter

Figure 4.1 (left) shows the typical TES biasing with a constant DC current, and the bias current $I_{in}(t)$ in this case is given by

$$I_{\rm in}(t) = I_0.$$
 (4.4)

The TES is considered to be biased by a constant voltage $V_0 (\approx I_0 R_s)$ as long as $R_s \ll R_{\text{TES}}$ is maintained. The Joule heating of the TES, $P_{\rm b}(t)$, then becomes

$$P_{\rm b}(t) = \frac{V_0^2}{R_0} = P_0. \tag{4.5}$$



Fig. 4.1 TES biasing with DC-current (left) and AC-current (right)

The input current to the SQUID, $I_{sq}(T)$, is then given by

$$I_{\rm sq}(t) = \frac{V_0}{R(t)}.$$
(4.6)

Fourier transforming this equation gives

$$\hat{I}_{\rm sq}(\omega) = V_0 \mathcal{F}\left[\frac{1}{R(\omega)}\right],\tag{4.7}$$

and from (4.3) we obtain

$$\frac{1}{R(t)} = \frac{1}{R_0 + \Delta R(t)}$$
(4.8)

$$\approx \frac{1}{R_0} \left(1 - \frac{\Delta R(t)}{R_0} \right). \tag{4.9}$$

The Fourier transformation of above is

$$\mathcal{F}\left[\frac{1}{R(\omega)}\right] \approx \mathcal{F}\left[\frac{1}{R_0}\left(1 - \frac{R_0}{\Delta R(t)}\right)\right] = \frac{1}{R_0}\delta(\omega) - \frac{\Delta \hat{R}(\omega)}{{R_0}^2},\tag{4.10}$$

where $\delta(\omega)$ is the Dirac delta function. (4.6) thus becomes

$$\hat{I}_{\rm sq}(\omega) = \frac{V_0}{R_0} \delta(\omega) - \frac{V_0}{{R_0}^2} \Delta \hat{R}(\omega).$$
(4.11)

 $\Delta \hat{R}(\omega)$ is generally confined to a frequency range, $-\omega_s < \omega < \omega_s$. Therefore, the power spectrum of the TES signal is confined around $\omega = 0$ as in Figure 4.2.

We now consider the TES biasing with a sinusoidal AC current (Figure 4.1 right). Suppose that the bias current in this case is given by

$$I_{\rm in}(t) = \sqrt{2I_0 \cos(\omega_0 t + \theta)},\tag{4.12}$$

where ω_0 is the carrier frequency. The Joule heating of the TES then becomes

$$P_{\rm b}(t) = \frac{2V_0^2}{R_0} \cos^2(\omega_0 t + \theta).$$
(4.13)



Fig. 4.3 The signal power spectrum of AC-biased TES

0

The time-averaged Joule heating is thus given by

 $-\omega_0$

$$\bar{P}_{\rm b}(t) = \frac{V_0^2}{R_0} = P_0, \tag{4.14}$$

 ω_0

which is identical to the Joule heating (4.5) in the DC-biased base. The SQUID input in this case is given by

$$I_{\rm sq}(t) = \frac{\sqrt{2}V_0 \cos(\omega_0 t + \theta)}{R(t)},$$
(4.15)

and its Fourier transformation is then given by

$$\hat{I}_{\rm sq}(\omega) = \sqrt{2} \left[\frac{{\rm e}^{i\theta}}{2} \left\{ \frac{V_0}{R_0} \delta(\omega - \omega_0) - \frac{V_0}{{R_0}^2} \Delta \hat{R}(\omega - \omega_0) \right\} + \frac{{\rm e}^{-i\theta}}{2} \left\{ \frac{V_0}{R_0} \delta(\omega + \omega_0) - \frac{V_0}{{R_0}^2} \Delta \hat{R}(\omega + \omega_0) \right\} \right].$$
(4.16)

Therefore, the power spectrum of the TES signal is split into two components where $\omega = \pm \omega_0$ as in Figure 4.3.

To retrieve the TES signal from the modulated signal (4.16), it needs to be demodulated. There are several methods for the demodulation. One way is a phase detection. By multiplying (4.15) with $\sqrt{2}\cos(\omega_0 t + \theta')$, we obtain

$$I'_{\rm sq}(t) = \sqrt{2}I_{\rm sq}(t)\cos(\omega_0 t + \theta') = \frac{2V_0}{R(t)}\cos(\omega_0 t + \theta)\cos(\omega_0 t + \theta'),$$
(4.17)



Fig. 4.4 The demodulated signal power spectrum using the phase detection method

and its Fourier transformation is given by

$$\hat{I}_{sq}'(\omega) = \cos(\theta - \theta') \left\{ \frac{V_0}{R_0} \delta(\omega) - \frac{V_0}{{R_0}^2} \Delta \hat{R}(\omega) \right\}
+ \frac{e^{i(\theta + \theta')}}{2} \left\{ \frac{V_0}{R_0} \delta(\omega - 2\omega_0) - \frac{V_0}{{R_0}^2} \Delta \hat{R}(\omega - 2\omega_0) \right\}
+ \frac{e^{-i(\theta + \theta')}}{2} \left\{ \frac{V_0}{R_0} \delta(\omega + 2\omega_0) - \frac{V_0}{{R_0}^2} \Delta \hat{R}(\omega + 2\omega_0) \right\}.$$
(4.18)

When $\theta = \theta'$, the power spectrum of the demodulated signal becomes as in Figure 4.4. We can retrieve the first term on the right-hand side by applying a low-pass filter and obtain

$$\hat{I}_{\rm sq}^{\prime\prime}(\omega) = \frac{V_0}{R_0} \delta(\omega) - \frac{V_0}{{R_0}^2} \Delta \hat{R}(\omega), \qquad (4.19)$$

which is exactly the same as (4.11).

Another method is an I/Q demodulation. In this method, we multiply (4.15) with $\sqrt{2}\sin(\omega_0 t + \theta')$ and $\sqrt{2}\cos(\omega_0 t + \theta')$, and obtain

$$I(t) = \frac{2V_0}{R(t)}\cos(\omega_0 t + \theta)\sin(\omega_0 t + \theta'), \qquad (4.20)$$

$$Q(t) = \frac{2V_0}{R(t)}\cos(\omega_0 t + \theta)\cos(\omega_0 t + \theta'), \qquad (4.21)$$

where I(t) and Q(t) are called in-phase and quadrature components, respectively. Their Fourier transformations are given by

$$\hat{I}(\omega) = \sin(\theta - \theta') \left\{ \frac{V_0}{R_0} \delta(\omega) - \frac{V_0}{R_0^2} \Delta \hat{R}(\omega) \right\} + \frac{e^{i(\theta - \theta')}}{2i} \left\{ \frac{V_0}{R_0} \delta(\omega - 2\omega_0) - \frac{V_0}{R_0^2} \Delta \hat{R}(\omega - 2\omega_0) \right\} + \frac{e^{-i(\theta - \theta')}}{2i} \left\{ \frac{V_0}{R_0} \delta(\omega + 2\omega_0) - \frac{V_0}{R_0^2} \Delta \hat{R}(\omega + 2\omega_0) \right\}$$
(4.22)



Fig. 4.5 Equivalent noise circuit for AC-biased microcalorimeter

for I(t) and

$$\hat{Q}(\omega) = \cos(\theta - \theta') \left\{ \frac{V_0}{R_0} \delta(\omega) - \frac{V_0}{{R_0}^2} \Delta \hat{R}(\omega) \right\} + \frac{e^{i(\theta + \theta')}}{2} \left\{ \frac{V_0}{R_0} \delta(\omega - 2\omega_0) - \frac{V_0}{{R_0}^2} \Delta \hat{R}(\omega - 2\omega_0) \right\} + \frac{e^{-i(\theta + \theta')}}{2} \left\{ \frac{V_0}{R_0} \delta(\omega + 2\omega_0) - \frac{V_0}{{R_0}^2} \Delta \hat{R}(\omega + 2\omega_0) \right\}$$
(4.23)

for Q(t). Again we retrieve the first term by applying a low-pass filter and obtain

$$\hat{I}'(\omega) = \sin(\theta - \theta') \left\{ \frac{V_0}{R_0} \delta(\omega) - \frac{V_0}{{R_0}^2} \Delta \hat{R}(\omega) \right\},$$
$$\hat{Q}'(\omega) = \cos(\theta - \theta') \left\{ \frac{V_0}{R_0} \delta(\omega) - \frac{V_0}{{R_0}^2} \Delta \hat{R}(\omega) \right\},$$

and by taking an absolute value of these vectors we finally obtain the demodulated signal as the same as (4.11). This method is superior to the phase detection method in a way that it is independent to θ' .

4.1.3 Noise of AC-biased microcalorimeters

Figure 4.5 shows the equivalent noise circuit for an AC-biased microcalorimeter. R, V, R_{in} , and A are the TES resistance, the bias voltage, the input impedance of ammeter, and the gain of readout system, respectively. r, i_R , and i_A are noise sources. r represents the noise that appears in the TES resistance, such as the phonon noise, and satisfies $R_0 \gg |\Delta R(t)| \gg |r(t)|$. i_R and i_A represent the current noise due to Johnson noise and the equivalent input voltage noise of the readout system, respectively.

Let us first consider the DC-biasing case using Figure 4.5. When $V(t) = V_0$, the TES output current (or SQUID input current) is given by

$$I_{\rm sq}(t) = \frac{V_0}{R(t) + r(t)} + i_R(t) + i_A(t)$$
(4.24)

$$\approx \frac{V_0}{R_0} \left(1 - \frac{\Delta R(t) + r(t)}{R_0} \right) + i_R(t) + i_A(t).$$
(4.25)

From (4.10), the Fourier transformation of above becomes

$$\hat{I}_{\rm sq}(\omega) \approx \frac{V_0}{R_0} \delta(\omega) - \frac{V_0}{{R_0}^2} \Delta \hat{R}(\omega) - \frac{V_0}{{R_0}^2} \hat{r}(\omega) + \hat{i}_R(\omega) + \hat{i}_A(\omega).$$
(4.26)

The first two terms at the right-hand side represent the signal on a X-ray event, and its power spectral density

is given by

$$\hat{i}_{\rm S}^2(\omega) = \frac{V_0^2}{R_0^4} \Delta \hat{R}^2(\omega).$$
(4.27)

The rest represents the noise, and its power spectral density is given by

$$\hat{i}_{N}^{2}(\omega) = \hat{i}_{N1}^{2}(\omega) + \hat{i}_{N2}^{2}(\omega), \qquad (4.28)$$

where

$$\hat{i}_{N1}^2(\omega) = \frac{V_0^2}{R_0^4} \hat{r}^2(\omega), \qquad (4.29)$$

which represents the noise due to the TES resistance, and

$$\hat{i}_{N2}^{2}(\omega) = \hat{i}_{R}^{2}(\omega) + \hat{i}_{A}^{2}(\omega), \qquad (4.30)$$

which represents the noise due to Johnson noise and the readout system noise. Therefore, the signal-to-noise ratio (SNR) for the DC-biasing case is

$$(SNR_1)^2 = \frac{\Delta \hat{R}^2(\omega)}{\hat{r}^2(\omega)},\tag{4.31}$$

$$(SNR_2)^2 = \frac{V_0^2}{R_0^4} \frac{\Delta \hat{R}^2(\omega)}{\hat{i}_R^2(\omega) + \hat{i}_A^2(\omega)},\tag{4.32}$$

where $SNR_1 \equiv i_{\rm S}/i_{\rm N1}$ and $SNR_2 \equiv i_{\rm S}/i_{\rm N2}$.

We now consider the AC-biasing case. Suppose that we have

$$V(t) = \sqrt{2}V_0 \cos(\omega_0 t).$$
(4.33)

The TES output (or SQUID input) is then given by

$$I_{\rm sq}(t) = \frac{\sqrt{2}V_0 \cos(\omega_0 t)}{R(t) + r(t)} + i_R(t) + i_A(t)$$
(4.34)

$$\approx \frac{\sqrt{2}V_0 \cos(\omega_0 t)}{R_0} \left(1 - \frac{\Delta R(t) + r(t)}{R_0} \right) + i_R(t) + i_A(t), \tag{4.35}$$

and its Fourier transformation is given by

$$\hat{I}_{sq}(\omega) = \sqrt{2} \left[\frac{1}{2} \left\{ \frac{V_0}{R_0} \delta(\omega - \omega_0) - \frac{V_0}{{R_0}^2} \Delta \hat{R}(\omega - \omega_0) \right\} + \frac{1}{2} \left\{ \frac{V_0}{R_0} \delta(\omega + \omega_0) - \frac{V_0}{{R_0}^2} \Delta \hat{R}(\omega + \omega_0) \right\} \right] - \sqrt{2} \left[\frac{1}{2} \frac{V_0}{{R_0}^2} \hat{r}(\omega - \omega_0) + \frac{1}{2} \frac{V_0}{{R_0}^2} \hat{r}(\omega + \omega_0) \right] + \hat{i}_R(\omega) + \hat{i}_A(\omega). \quad (4.36)$$

Let us demodulate (4.35) and apply a low-pass filter, then we obtain

$$\hat{I}'_{\rm sq}(\omega) = \left(\frac{V_0}{R_0}\delta(\omega) - \frac{V_0}{{R_0}^2}\Delta\hat{R}(\omega) - \frac{V_0}{{R_0}^2}\Delta\hat{r}(\omega)\right) \\
+ \frac{\sqrt{2}}{2}\left\{\hat{i}_R(\omega - \omega_0) + \hat{i}_R(\omega + \omega_0) + \hat{i}_A(\omega - \omega_0) + \hat{i}_A(\omega + \omega_0)\right\}.$$
(4.37)


Fig. 4.6 The schematic diagram of the baseband feedback architecture. Only the single channel configuration is shown.

Therefore, the power spectral density of noise becomes

$$\hat{i}_{\rm N1}^2(\omega) = \frac{V_0^2}{R_0^4} \hat{r}^2(\omega), \qquad (4.38)$$

which remains the same, and

$$\hat{i}_{N2}^{2}(\omega) = \frac{1}{2} \left\{ \hat{i}_{R}^{2}(\omega - \omega_{0}) + \hat{i}_{R}^{2}(\omega + \omega_{0}) + \hat{i}_{A}^{2}(\omega - \omega_{0}) + \hat{i}_{A}^{2}(\omega + \omega_{0}) \right\}.$$
(4.39)

When $\hat{i}_R^2(\omega - \omega_0) = \hat{i}_R^2(\omega + \omega_0)$ and $\hat{i}_A^2(\omega - \omega_0) = \hat{i}_A^2(\omega + \omega_0)$ within the signal band $-\omega_s < \omega < \omega_s$, the above equation can be given as

$$\hat{i}_{N2}^2(\omega) = \hat{i}_R^2(\omega_0) + \hat{i}_A^2(\omega_0), \qquad (4.40)$$

and the SNR becomes identical to (4.31) and (4.32). This means that the difference of SNR between the DCbiasing case and the AC-biasing case is determined only by (4.30) and (4.40). The readout noise at low frequency is generally larger on account of 1/f noise, and the SNR in AC-biasing is thus larger than that in DC-biasing.

4.1.4 Principles of Baseband Feedback

In FDM TES are typically biased at several MHz, which is sufficiently faster than the thermal time constant of TES, and is sufficiently large to multiplex enough number of TES channels. However, it is too fast to stably feedback in the flux-locked loop mostly because of the parasitic capacitance of electrical wires between the cryogenic stage and the room temperature electronics. The typical parasitic capacitance of a twisted-pair of wires used in a refrigerator is $\sim 100 \,\mathrm{pF/m}$, and the typical length of the wires is $\sim 1 \,\mathrm{m}$ or more. As a result, the parasitic capacitance can be several hundreds of pF, which is critically high for the FLL in MHz. To achieve a stable feedback even in MHz frequency, a feedback scheme called baseband feedback (BBFB) is used.

Figure 4.6 shows the schematic diagram of BBFB. The modulated signal is first demodulated at the phase sensitive detector (or the I/Q demodulator) and the baseband signal is extracted. The integrator is then applied to the baseband signal, as is done in the DC-biasing FLL. Finally, another carrier signal that is phase-adjusted, so that the phase becomes opposite at the summing point, is modulated by the baseband signal, and then sent back to the SQUID to nullify the input. In this way, we succeeded in the stable FLL for a 5 MHz carrier [30]. Although BBFB overcomes the wire delay, the closed loop of FLL should still be as short as possible to eliminate

	Minimal (8 multiplexing)	Goal (16 multiplexing)
Bandwidth	$> 1.5\mathrm{MHz}$	$> 3 \mathrm{MHz}$
ADC SNR (and theoretical bit width)	60 dB (10 bits)	60 dB (10 bits)
DAC SNR (and theoretical bit width)	78 dB (13 bits)	84 dB (14 bits)
ADC and DAC Sampling Rate	$\gg 6\mathrm{MHz}$	$\gg 12\mathrm{MHz}$
FPGA Multiplier	> 200	> 400
FPGA RAM	$> 4 \mathrm{MB}$	$> 8 \mathrm{MB}$

Table 4.1 Requirements for digital electronics

the baseband signal delay, which makes a finite input into SQUID even in the FLL and ultimately causes a flux jump.

In our previous study, we succeeded to multiplex two TES signals, one biased with a 1.0 MHz carrier and the other biased with a 1.5 MHz carrier, and to readout X-ray pulses simultaneously [34]. We used the analog-based room-temperature electronics for BBFB, which jointly-developed with NF corporation, but it supports only up to 2 MHz carrier frequency, thus it is not suitable for high-density multiplexing. Moreover, the lack of scalability in analog-based circuits may become the major drawback when realizing a readout system for large-format TES arrays. As we have successfully demonstrated the principle of BBFB using the analog electronics, we started to develop a scalable FPGA-based digital readout system.

4.2 Requirements for Digital Electronics

To multiplex 256 to 400 TES signals with a reasonable number of SQUIDs, a single SQUID should multiplex at least 8 signals, and preferably 16 signals or more. A typical bandwidth of a baseband signal (TES signal) is ~ 100 kHz, hence a required spacing frequency between multiplexed channels is ≥ 200 kHz, which makes the number of multiplexed channels within 1 MHz bandwidth to be 5 or less. To multiplex at least 8 signals with a single SQUID, we need a bandwidth of more than 1.5 MHz.

The TES for DIOS will be optimized for soft X-rays ($\leq 2 \text{ keV}$) and the requirement on energy resolution is 2 eV, so the required signal-to-noise ratio (SNR) for the system is at least 60 dB. In FDM with BBFB, the SQUID output will be suppressed by a factor of the loopgain, which is the function of baseband frequency. The lower frequency (~DC) of the signal gets almost fully suppressed, but the higher frequency may not get suppressed. However, it would cause an excess only when photons hit TES at the very same time, which should rarely occur for astronomical observations where the typical count rate is less than 1 cps. Therefore, an increase of multiplexing channels would not cause an excess in ADC dynamic ranges. On the other hand, as the multiplexing number increases, a dynamic range of DAC for a single channel decreases, so the required SNR for DAC would be higher than that of ADC, and can be expressed as $60 + 6.02 \times \log_2$ (number of multiplexing channels) dB. To multiplex 8 signals, the required SNR for DAC would be 78 dB, while it would be 84 dB to multiplex 16 signals. Therefore, a 13-bit DAC is sufficient to multiplex 8 signals, while a 14-bit DAC is sufficient to multiplex 16 signals theoretically. In a real world scenario we need more DAC bits due to noises. Obviously both ADC and DAC need to have sufficient sampling rates to satisfy the required system bandwidth. To multiplex 8 signals, the required bandwidth is > 1.5 MHz, and if a band from 1.5 to 3.0 MHz is used, then we need a sampling rate of more than 6 MHz. To multiplex 16 signals, we need a doubled sampling rate.

Recent fast field-programmable gate arrays (FPGAs) have plenty of logic cells, block RAMs, and multipliers, so we may not need to worry about resources when selecting a FPGA. It should have at least several hundreds of multipliers (at least 16-bit×16-bit) for signal processing, and several megabytes of RAM for data acquisition.

We summarized the requirements for digital electronics in Table 4.1.

Table 4.2 FINCT50 specifications			
ADC (ADS62P49)			
Number of channels	2		
Channel resolution	14-bit		
Input voltage range	$2 \mathrm{V_{pp}} \ (10 \mathrm{dBm})$		
Input gain	Programmable from 0 dB to 6 dB in 0.5 dB steps		
Input impedance	50Ω (AC-coupled)		
Analog input bandwidth	$0.40-500~\mathrm{MHz}$		
SNR	$71~\mathrm{dBFS} @~45~\mathrm{MHz}$		
SFDR	$80~\mathrm{dBc} @~45~\mathrm{MHz}$		
DAC (DAC3283)			
Number of channels	2		
Channel resolution	16-bit		
Output voltage range	$1 \mathrm{V_{pp}}$		
Output impedance	50Ω (AC-coupled)		
Analog output bandwidth	82 MHz 5th-order Chebyshev LPF		
	3 MHz HPF due to the output transformer		
THD	$-67\mathrm{dBc}$		

Table 4.2 FMC150 specification

4.3 System Overview

For rapid prototyping, this time we selected off-the-shelf FPGA and ADC/DAC boards.

4.3.1 ADC/DAC Board

We selected the 4DSP FMC150 ADC/DAC board. It has two 250Msps 14-bit ADC and two 800Msps 16-bit DAC, and can be mounted to FPGA boards with a low-pin count (LPC) FPGA mezzanine card (FMC) connector. It is based on Texas Instruments (TI) ADS62P49 and TI DAC3283. The detailed specifications are summarized in Table 4.2. The input bandwidth is broad enough to use for the BBFB, while the output bandwidth has a low-cut off at 3 MHz, which may causes a loss of loopgain in smaller frequency. The loopgain, however, can be adjusted in each channel, and the effect of the low-cut off thus may be negligible.

The input voltage range of the ADC is $2 V_{pp}$, while its of the DAC is $1 V_{pp}$. To adjust the input/output range, we configured the ADC input gain to be 6 dB. This makes the actual input bit width to be 13 bits.

The DAC has internal $\times 2$ and $\times 4$ interpolators, however, we do not use any of them as they introduce additional delays in signals (discussed in Section 4.5).

4.3.2 FPGA Board

We selected the Xilinx ML605 evaluation board^{*1}. It is based on Xilinx Virtex-6 LX240T (XC6VLX240T-1FFG1156) FPGA, which provides 240k logic cells, 768 25-bit×18-bit multipliers, and 15 MB block RAM. The board has one LPC FMC connector and one high-pin count (HPC) FMC connector, so it can mount up to two FMC150 boards, which makes possible to multiplex 32 signals thanks to the plenty of FPGA resources.

 $^{^{\}ast 1}$ The system is also ported to the Xilinx KC705 evaluation board.



Fig. 4.7 Digital BBFB diagram

4.4 Design and Implementation

4.4.1 Core Logic

Figure 4.7 shows the overall block diagram (single-channel unit except ADC, DAC and FIFO) of a developed digital BBFB system. The incoming signal is digitized in 245.76 Msps by the ADC, and then spread to each channel unit. In analog electronics, a phase sensitive detector was used to demodulate, but we used an I/Q demodulator for a stable demodulation. The demodulated I and Q signals are decimated by 64 and down-sampled to 3.84 Msps. To filter out second harmonic generated at the I/Q demodulator, and to FLL, I and Q signals are integrated by the discrete-time integrator. To generate the feedback signal, both signals are interpolated by 64, and then up-converted to the carrier frequency by the I/Q modulator. The feedback signal is then summed up with other channel feedback signals, divided by the number of multiplexing channels, and finally converted to the analog signal by the DAC in 245.76 Msps.

The baseband signal, which is the absolute value of I and Q signals, is calculated by the coordinate rotation digital computer (CORDIC) algorithm. The baseband signal optionally flipped upside down, and then applied the 3rd-order low-pass filter. The signal is still just a digital stream, and to extract a pulse (and a noise) waveform, the signal is triggered by a threshold trigger, extracted preset number of points, and then transfered to the first-in-first-out (FIFO) buffer.

Here we elaborate more on each component.

I/Q Demodulator and Modulator

The I/Q demodulator takes the input signal data stream of 16-bit signed integer from the ADC, and sine/cosine data streams of 16-bit signed integer from a direct digital synthesizer (DDS), and multiplies them to produce the I and Q signals. Two least significant bits (LSB) of input signal is dummy (zero), since the ADC is 14-bit converter. The DDS is generated using the Xilinx DDS Compiler (ver. 4.0). The frequency and phase of the



Fig. 4.8 CIC Decimator

DDS output are programmable, and the bit width of the frequency/phase register is 18 bits, which makes the frequency resolution to be ~ 1 kHz (245.76 MHz/2¹⁸), and the phase resolution to be ~ 2.4×10^{-5} rad. The output bit width of the DDS is 16 bits, and therefore the spurious free dynamic range (SFDR) is 96 dB. The input signal and the sine/cosine signals are multiplied with the pipelined multiplier (3 stages) clocked in 245.74 MHz using the DSP48E1 hardware multiplier. The bit width of the product is 32 bits, and we extract 16 bits from 2nd most significant bits (MSB) as the I and Q signals, since the two MSB are both signed bits (unless both signals are -32768), which makes the gain at this stage to be 1/2 if we cut off the higher frequency component. The sine signal from the DDS is also output for the carrier signal.

The I/Q modulator takes the I and Q input signal data stream of 16-bit signed integer, and sine/cosine data streams of 16-bit signed integer from another DDS, multiplies them, and finally sums up to produce the feedback signal. The DDS is identical to the I/Q demodulator DDS except that only the frequency is programmable. The output is extracted in the same manner.

The carrier signal and the feedback signal are summed up with other channel carrier signals and feedback signals, divided (arithmetic right shift) by the number of channels, or the output scaler, and transferred to the DAC.

CIC Decimator and Interpolator

The signal sampling rate, 245.76 Msps, is too fast to process for demodulated signals with a small bandwidth (< 1 MHz), and therefore is down-sampled to a much smaller frequency, 3.84 Msps. The I and Q signals from the I/Q demodulator are decimated by 64 using a cascaded integrator-comb (CIC) filter. The CIC filter is efficient way to implement a combination of a simple boxcar filter (averaging filter) and a simple decimator (or simply a moving-average filter) [15]. It consists of three stages; an integrator stage, a 1/N decimator, and a comb stage (Figure 4.8).

The transfer function of the CIC decimation filter is given by

$$H(z) = \left(\sum_{k=0}^{RM-1} z^{-k}\right)^{N}$$
(4.41)

$$= \left(\frac{1 - z^{-RM}}{1 - z^{-1}}\right)^{N}, \tag{4.42}$$

where R is the decimation (or interpolation) ratio, M is the differential delay, which is usually 1 or 2, and N is the number of stages in the filter. Figure 4.9 shows the frequency responses of the 3rd-order 1/64 CIC decimation filter for both M = 1 and M = 2 cases. In both cases, integral multiplications of f_s/R , where f_s is



Fig. 4.9 Frequency responses of 3rd-order 1/64 CIC decimation filter

the sampling frequency, becomes zero, which suppresses the aliased spectral components due to the decimation (or interpolation). Therefore, the CIC filter is well-suited for anti-aliasing filtering.

The properly decimated signal is known to have a larger effective number of bits (ENOB), because of the improved SNR. The growth of ENOB, Δ ENOB, after the decimation with the decimation rate R is given by

$$\Delta \text{ENOB} = \log_4 R, \tag{4.43}$$

therefore the ENOB increases by 3 bits when R = 64. As we are using the 3rd-order differential-delayed 1/64 decimation filter for the I and Q signals, the bit width after the decimation becomes 19 ignoring the fact that two LSB of incoming signals are dummy and that the actual ENOB of incoming signals are smaller than 14 due to noise. The decimation rate is also configurable to be 128, so we are making the bit width to be 20.

The CIC interpolator takes the integrated I and Q signal data stream of 16-bit signed integer clocked in 3.84 MHz and up-samples to 245.76 Msps. The output bit width remains 16.

Discrete-Time Integrator and Gain Control

The integrator is the key component for the FLL. In analog electronics, the integrator is a single op-amp integrator for a single channel, while it is a set of two discrete-time integrators for each I and Q signals in digital electronics. The integrator is actually a simple infinite impulse response (IIR) filter which transfer function is given by

$$H(z) = \frac{a}{1 - z^{-1}},\tag{4.44}$$

where a is the gain parameter of the integrator. The gain parameter is programmable from 2^{-4} to 2^3 in 8 steps using a barrel shifter. Figure 4.10 shows the frequency responses of the discrete-time integrator for $a = 2^{-4}$, 2^{-2} , 2^0 and 2^2 cases. Unlike the analog integrator which DC gain is limited by the op-amp open-loop gain, the DC gain of the discrete-time integrator is infinity, and the carrier signal is thus almost completely suppressed (but limited by the ADC effective resolution) in the FLL loop.

The bit width of incoming I and Q signals to the gain control component is 20, and the gain control attenuates signals by 2^4 at a maximum, therefore the output bit width from the gain control is 24. The integrator then takes 24-bit signed integer data streams, integrates them, and outputs them as 16-bit signed integer data stream taking the 16 most significant bits.



Fig. 4.10 Frequency responses of discrete-time integrator for various gains



Fig. 4.11 Biquad filter implementations: direct form 1 (left) and direct form 2 (right)

CORDIC

To calculate the absolute value of the I and Q vectors, the CORDIC algorithm is used. It is a calculation technique to break trigonometric calculations to simple additions, subtractions, and bit shifts. The CORDIC is generated using the Xilinx CORDIC (ver. 4.0). It takes I and Q 16-bit signed integer data streams, calculates their absolute values, and output them as 16-bit unsigned integer data stream. Upon the calculation, the I and Q signals are normalized to be from $-1 \leq (I \text{ or } Q) \leq 1$, then the absolute value range becomes from 0 to 1, and finally it is scaled so that the absolute value 0 to be 0 and 1 to be $2^{16} - 1$.

3rd-order LPF for Baseband

The demodulated signal is filtered by the CIC decimator and the integrator, yet it may still include signals from neighbor channels. To filter out those undesired signals, the 3rd-order low-pass filter (LPF) is applied to the demodulated signal. The filter consists of cascaded IIR biquad filter (direct form 1). Figure 4.11 shows the two types of biquad filter implementations, the direct form 1 and the direct form 2. The transfer function of both types is the same, and is given by

$$H(z) = \frac{b_0 + b_1 z^{-1} + b_2 z^{-2}}{1 + a_1 z^{-1} + a_2 z^{-2}}.$$
(4.45)

The direct form 2 is usually preferred since it uses less amount of delays, however, we used the direct form 1 that maximize the merit of the multiply accumulator (MACC) of the FPGA DSP core.

The filter can be designed to be many types of signal filters, including high-pass filters, band-pass filter, and so

Table 4.9 Daseballa El 1 de	sign parameters
Filter type	Butterworth
Passband	$1/64 \ (\pi \ rad/sample) \ (30 \rm kHz)$
Stopband	$1/16 \ (\pi \ rad/sample) \ (120 \rm kHz)$
The maximum loss in the passband	3 (dB)
The minimum attenuation in the stopband	20 (dB)

Table 4.3 Baseband LPF design parameters



Fig. 4.12 Frequency responses of biquad low-pass filter for various fraction bits

on. It can be also designed to be several types of LPF, such as Chebyshev filters and elliptic filters. We designed our biquad low-pass filter to be a simple Butterworth filter as in Table 4.3, and the coefficients are

$$b_0 = 0.00093078, \tag{4.46}$$

$$b_1 = 0.00186157, \tag{4.47}$$

$$b_2 = 0.00093078, \tag{4.48}$$

$$a_1 = -1.91177368, \tag{4.49}$$

$$a_2 = 0.91549682. \tag{4.50}$$

To implement the filter in digital, the coefficients need to be quantized. Figure 4.12 shows the frequency responses of the filter for various fraction bits. The smaller bits cause inconsistencies from the ideal transfer function, therefore we used 16 bits for fraction bits.

Figure 4.13 shows the frequency response of the 3rd-order low-pass filter for the baseband signal. Both input and output signals to the filter are 16-bit unsigned integers.

Trigger, Baseband Decimation and Waveform Extract

The baseband signal is still the data stream, and pulses (and noises) need to be triggered and extracted. The trigger is a simple threshold trigger. The trigger direction can be configured in both directions.

The signal stream through the trigger can be changed to the 1st-order derivative or 2nd-order (pseudo-) derivative of the 2-point or 4-point averaged signal. The 2-point 1st-order derivative filter is a convolution of a 2-point moving average filter and a simple derivative filter which transfer function is given by

$$H(z) = 1 - z^{-1}, (4.51)$$



Fig. 4.13 Frequency response of 3rd-order low-pass filter

and its transfer function is given by

$$H(z) = 0.5 - 0.5z^{-2}.$$
(4.52)

Similarly the transfer function of 4-point 1st-order derivative filter is given by

$$H(z) = 0.25 - 0.25z^{-4}.$$
(4.53)

The 2nd-order derivative filters are a convolution of moving average filters and a 2nd-order pseudo-derivative filter which transfer function is given by

$$H(z) = 1 - z^{-1} - z^{-2} + z^{-3}, (4.54)$$

while that of ideal 2nd-order derivative filter is given by

$$H(z) = 1 - 2z^{-1} + z^{-2}.$$
(4.55)

The reason for the pseudo-derivative filter is to reduce the number of calculations. The transfer functions of 2-point and 4-point 2nd-order pseudo-derivative filters are given by

$$H(z) = 0.5 - z^{-2} + 0.5z^{-4} \qquad (2 \text{ points}), \tag{4.56}$$

$$H(z) = 0.25 - 0.25z^{-2} - 0.25z^{-4} + 0.25z^{-6} \qquad (4 \text{ points}), \tag{4.57}$$

respectively. However, to further reduce the number of calculations for (4.57), we merged the second and the third terms and obtained

$$H(z) = 0.25 - 0.5z^{-3} + 0.25z^{-6}.$$

The sampling rate of the baseband signal after the low-pass filter is 3.84 Msps. However, the effective signal bandwidth is far smaller. To reduce the data rate, we further decimate the baseband signal by 16, and the sampling rate after the decimation becomes 240 ksps which Nyquist frequency is the same as the stopband of the



Fig. 4.14 Dispersion of pulse heights (PHA) vs. sampling rates for various pulse rise times

low-pass filter. However, the trigger has to be applied before the decimation. Figure 4.14 shows the simulation results of a dispersion of pulse heights as a function of sampling rate for various pulse rise times. 10,000 noiseless pulses which pulse height is fixed to 1.0 but rise timings are varied within a single sampling clock were simulated, and then the maximum values for each pulse were taken as pulse heights. The fall time and the length of pulses are fixed to 200 μ s and 2 ms, respectively. Δ PHA is calculated as Δ PHA = |PHA_{ideal} – PHA|, where PHA_{ideal} = 1.0. To achieve the required energy resolution (2 eV @ 2 keV), the dispersion needs to be suppressed less than 10⁻³, therefore the sampling rate for triggering should be 1 Msps or more. For this reason, the decimation needs to be done after triggering.

The data stream of the baseband signal from the trigger component passes through the delay component, which delays the signal by a given amount of 3.84 MHz clocks for a positive trigger position, and then gets decimated by 16. The decimator is always reseted upon the trigger signal to align the extracted pulses. The bit width increases by 2 bits here, and the following gain control aligns the width to 16 bits. Finally a pulse (or noise) waveform is extracted into a local buffer from the signal stream with a given amount of data length chosen from 256, 512, 1024 or 2048 points, which corresponds to 1, 2, 4 or 8 ms length, and then flushed to the FIFO. A very little time during the data flush from the local buffer to the FIFO becomes a dead time. The flush is done in the 245.76 MHz clock, so the maximum dead time in the worst case is (data lengh)/245.76 MHz × (number of channels – 1), and is ~ 3 μ s for the 16-channel multiplexing 512-point configuration. If the dead time is critical, it can be completely avoided by adding another local buffer to do double buffering, yet it obviously requires more hardware resources.

The data sizes for each data length, 256, 512, 1024 and 2048 points, are 512, 1024, 2048 and 4096 bytes, respectively. The data rate is therefore ~ 500 kB/s per channel at maximum, and the system overall data rate for 16-channel multiplexing is ~ 8 MB/s at maximum. However, the required maximum count rate is far less in astronomical applications. If we assume the required count rate to be 100 cps, which is still very large, the data rate per channel becomes ~ 100 kB/s, and the system data rate for 16-channel multiplexing therefore becomes ~ 1.6 MB/s, which is reasonably small even for resource-limited satellite applications.

4.4.2 System Overall Gain

The overall gain in the flux-locked loop in this digital system is the sum of gains of the I/Q demodulator, the CIC decimator, the gain control, the integrator, the CIC interpolator, the I/Q modulator, and the signal divider right before the DAC. The gains at each stage is summarized in Table 4.4. Figure 4.15 shows the overall frequency

Stage		Gain	DC (Carrier) Gain
I/Q Demodulator	$G_{\rm demod}$	1/2	1/2
CIC Decimator	$G_{\rm dec}$	$ H_{ m dec}(\omega) $	1
Gain Control	G_{gainc}	1/16, 1/8, 1/4, 1/2, 1, 2, 4, or 8 (programmable)	\leftarrow
Integrator	$\bar{G}_{ m int}$	$ H_{ m int}(\omega) $	∞
CIC Interpolator	G_{interp}	$ H_{ m interp}(\omega) $	1
I/Q Modulator	G_{mod}	1	1
Signal Divider	$G_{\rm div}$	1/16, 1/8, 1/4, 1/2, or 1 (programmable)	\leftarrow

Table 4.4 Gains at each FLL stage



Fig. 4.15 Overall gain in the digital electronics for various gain parameters $(G_{div} = 1)$

response for various G_{gainc} values. As we see in the figure, the overall gain is well-determined by the integrator.

4.4.3 Interface

Extracted pulse and noise waveforms in the FIFO need to be transferred to a host machine. There are several ways to achieve it, and we use a system-on-chip (SOC) design with the LEON processor. The LEON processor was originally developed by the European Space Agency (ESA), and now is developed and supported by Gaisler Research. It is a 32-bit CPU microprocessor core, based on the SPARC-V8 RISC architecture and instruction set. Figure 4.16 shows our SoC design.

The custom module, the BBFB controller, is attached to the advanced peripheral bus (APB), which is bridged to the advanced high-performance bus (AHB) at the AHB/APB bridge and then connected to the processor (Figure 4.17). The core logic extracts pulse and noise waveforms and transfer them to the FIFO in the BBFB controller. Once waveform data is available in the FIFO, the controller generates an interrupt and demands the processor to pull the waveform data. The hardware registers for configuration parameters are memory-mapped in the 32-bit physical address space. Table 4.5 shows the excerpted memory-mapped hardware configuration parameters. The complete list of the parameters and registers is in Appendix B.

The SoC runs Linux (2.6) as an operating system. Figure 4.18 show the application stack. As the processor is interrupted, the data in the hardware FIFO is pulled to the software FIFO in the kernel driver. The waveform data can be accessed from the user space through the device file, and configuration registers can be accessed through the sysfs. The measured throughput from the hardware FIFO to the user space is $\sim 11 \text{ MB/s}$, which is large enough for the estimated required system data rate, 1.6 MB/s.



Fig. 4.16 SoC design with LEON3 microprocessor



Fig. 4.17 The BBFB controller attached to the APB

Remote clients on the same network (or even other networks) can access waveforms and registers by XML-RPC connecting a server running on the user space. Figure 4.19 shows the developed client software, the TES Workbench. It connects to the server, which is running on the FPGA, pulls waveforms, and saves them to a database file on a local machine. It can also control all the hardware parameters from the GUI.

4.4.4 Resource Utilization

Figure 4.20 shows the resource usage of the FPGA. It shows the usage only by the core logic. The block RAM/FIFO usage is higher even for the smaller number of channels due to the 64 kB hardware FIFO. Since the resource usage for the LEON3 SoC is fairly small (Table 4.6), we can implement up to 32 channels within the

Parameter Name	Register Name	R/W	Bit width
Device	Control Parameters		
Reset	Device Control	WO	1
IRQ Enable		RW	1
Record Length		RW	2
Trigger Delay		RW	8
FIFO Count	Device Statistics	RO	16
Previous Trigger	Trigger Threshold	RW	16
Next Trigger		RW	16
FIFO Data	FIFO	RO	32
Channel	Control Parameters		
Enable	Channel Control 1	RW	1
Clear		RW	1
Feedback Enable		RW	1
Integrator Enable		RW	1
Trigger-Level/Count Select		RW	1
Frequency/Phase Select		RW	1
Baseband Gain		RW	2
Modulation Enable		RW	1
Derivation Mode		RW	2
Trigger Mode		RW	4
Gain		RW	3
Amplitude		RW	12
Frequency/Phase	Channel Control 2	RW	16
Trigger-Level/Count		RW	16
Trigger Count	Channel Statistics	RO	16

Table 4.5 $\,$ Excerpted hardware configuration parameters. The complete list of parameters and addresses is in Appendix B.



Fig. 4.18 Application stack on the SoC

	Table 4.6	The FPGA	resource	usage by	v the	LEON3	SoC
--	-----------	----------	----------	----------	-------	-------	-----

	Usage
Slice Registers	4%
Slice LUTs	11%
Block RAM/FIFO	4%
DSP48E1	1%



Server View

Digital Electronics Configuration

Histogram View

Fig. 4.19 TES Workbench



Fig. 4.20 Resource utilization for various number of channels



Feedback

Fig. 4.21 Loopback mode



Fig. 4.22 $\,$ Dummy pulse used to modulate in the loopback test $\,$

single FPGA.

4.5 Loopback Test

We evaluated the developed digital electronics in a loopback mode (Figure 4.21). In the loopback mode, the DAC outputs are directly connected to the ADC inputs, and the carrier and the feedback are individually sampled by the ADC, then digitally summed.

4.5.1 SNR Evaluation

The TES bias for the target channel is modulated by a 2-ms dummy pulse (2 μ s rise time, 200 μ s fall time, and 80% modulation) shown in Figure 4.22, while other channels are not modulated. We used carrier frequencies from 5 to 7 MHz with 128 kHz spacing frequency.

The SNR, which is given by $\Delta PHA/PHA$, was calculated from the result of optimal filtering with 16,000



Fig. 4.23 The obtained signal-to-noise ratio in the loopback test



Table 4.7 FMC150 delay caused by the internal interpolation

Fig. 4.24 The measured and theoretical frequency responses in the single channel loopback mode for various gain parameters

pulses and noises (Figure 4.23). The obtained SNR for 16-channel multiplexing is ~ 72 dB achieving the required SNR (> 60 dB). Figure 4.23 also shows theoretical SNRs of ADC and DAC, and infers that the system SNR is limited by the DAC SNR for higher-density multiplexing as expected. If a higher SNR is needed, we can increase the number of DAC to decrease the number of channels sharing a single DAC channel to output carriers and feedbacks.

4.5.2 Loopgain Evaluation

We first measured the delay caused by the FMC150. Table 4.7 shows the delay caused by the internal interpolation of the DAC. When $\times 4$ interpolation is enabled, the delay is almost doubled. As the sampling rate for both ADC and DAC is 245.76 Msps that is much higher than the signal frequency (< 10 MHz), therefore we do not use the interpolation. We then measured the frequency responses (amplitude and phase).



Fig. 4.25 The averaged pulses (left) and the averaged noises (right) for various spacing frequencies

Figure 4.24 shows the measured frequency responses of demodulated signals in the single channel loopback mode for various gain parameters along with the theoretical responses. While they show good agreement in magnitude, the measured phases roll off in smaller frequency than the theoretical phases, meaning that there is further delay in the system. The horizontal dashed line shows where the phase is -180° , and the gain margins are therefore -26.6 dB, -14.6 dB, and -2.6 dB for $G_{\text{gainc}} = 1/64$, 1/16 and 1/4 in $G_{\text{div}} = 1$, respectively. The unity-gain bandwidths are 4.3 kHz, 17.1 kHz and 66.8 kHz, for $G_{\text{gainc}} = 1/64$, 1/16 and 1/4, respectively, while the phase margin is $\sim 60^{\circ}$ for all the cases, which still leaves enough room for other delays.

4.5.3 Crosstalk Evaluation

In four channel multiplexing mode, we setup three channels with carrier frequencies, 5.25, 5.50, 5.75 MHz, and modulated all the channels with the dummy pulse. We collected pulses and noises for the middle channel, and calculated the averaged pulse. Next we changed the spacing frequency from 250 kHz to 125 kHz, therefore changed the carrier frequencies to 5.375, 5.500, 5.625 MHz, and collected the pulses and noises for the middle channel to calculate the averaged pulse. Figure 4.25 shows the obtained averaged pulses and averaged noises for the spacing frequencies 250 kHz and 125 kHz. We clearly see crosstalk for the averaged pulse with $f_s = 125$ kHz from the neighbor channels. The average noise of $f_s = 125$ kHz is also considerably higher. The gain control was set to $G_{\text{gainc}} = 1/2$ while the signal divider was set to $G_{\text{div}} = 1/4$. It makes the equivalent gain to be 1/8 in the single channel mode, and the unity-gain bandwidth is therefore ~ 35 kHz. Despite the frequency margin of ~ 50 kHz even for the 125 kHz spacing frequency, the crosstalk is not negligible.

We thus varied the gain control and collected pulses and noises for the $f_s = 125$ kHz case. Figure 4.26 shows the averaged pulses (left) and the averaged noise (right) for $G_{\text{gainc}} = 1/2$, 1/4, 1/8 and 1/16. Although the crosstalk becomes less noticeable when $G_{\text{gainc}} \leq 1/4$ for averaged pulse, the noise level for $G_{\text{gainc}} = 1/4$ is still higher. For $G_{\text{gainc}} \leq 1/8$, the noise drops to the usual level.

The crosstalk, of course, happens when more than two photons hit adjoining TES channels almost simultaneously, and it should be very rare in astronomical applications where the typical count rate is very small. However, when the count rate is expected to be very high, the spacing frequency has to be sufficiently large, otherwise the loopgain needs to be undesirably small.



Fig. 4.26 The averaged pulses (left) and the averaged noises (right) for various gain parameters ($f_s = 125 \text{ kHz}, G_{\text{div}} = 1/4$)



Fig. 4.27 A typical experimental setup for a TES multiplexed readout

4.6 Development of Analog Front-End

4.6.1 Requirements for Analog Front-End

The output voltage range of the DAC on FMC150, $1 V_{pp}$, is too small to use as a voltage source for TES biasing and the feedback. Figure 4.27 shows a typical experimental setup for a TES readout using the developed digital system. The typical Ti/Au TES operating resistance, R_{TES} , is ~ 50 m Ω , while the TES shunt resistance, R_S , is ~ 5 m Ω . This makes the separation ratio of the TES bias current, I_{TB} , to be $I_{TES} : I_S \approx 1 : 10$, where I_{TES} is the TES current, and I_S is the shunt current. The typical TES current at the operating point is ~ 50 μ A_{rms}, meaning that the TES bias current needs to be 500 μ A_{rms} or larger for single channel biasing. Suppose that we multiplex 10 TES signals, then I_{TB} can be as large as 5 mA_{rms}. To make such a current flow with the 1 V_{pp} voltage source, the bias resistance becomes ~ 70 Ω , which is smaller than the typical wire resistance, ~ 100-200 Ω , between the cryogenic stage and the room temperature electronics. In other words, the output impedance of the current

Low-Noise Amp			
Gain	$46\mathrm{dB}$		
Bandwidth	m DC-20MHz		
Equivalent input voltage noise	$< 2\mathrm{nV}/\sqrt{\mathrm{Hz}}@1\mathrm{MHz}$		
Input bias current	$< 1 \mathrm{nA}$ (abs.)		
Input impedance	$1 \mathrm{M}\Omega$ @DC		
Output impedance	50Ω		
V/I Converter			
Output impedance	$> 10 \mathrm{k}\Omega@1 \mathrm{MHz}$		

Table 4.8 Requirements for analog front-end



Fig. 4.28 The circuit schematic of typical FET differential amplifier

source is too small.

For a low-noise amp, we have been using ultra-low-noise FET-input differential preamplifier SA-421F5 by NF corporation. It has the very small equivalent input voltage noise of $0.5 \text{ nV}/\sqrt{\text{Hz}}$ (> 100 kHz), and the very small equivalent input current noise of $100 \text{ fA}/\sqrt{\text{Hz}}$, thus is well suited for SQUID readout. Moreover, it has the very wide bandwidth of 30 to 30 MHz, thus can be used for FDM. However, its input is AC-coupled, and has therefore no sensitivity to very slow signal variations, such as temperature drift.

For those reasons, we started to develop an analog front-end which is inserted in between the cryogenic electronics and the digital system. The requirements for the analog front-end are summarized in Table 4.8. Since our SQUID is biased with a very small current ($< 20 \,\mu$ A), the input bias current of low-noise amp needs to be as small as possible, preferably less than 1 nA. The typical TES noise, $20 \,\text{pA}/\sqrt{\text{Hz}}$ and the typical SQUID gain, $100 \,\text{V/A}$, make the SQUID output voltage noise to be $2 \,\text{nV}/\sqrt{\text{Hz}}$. The equivalent input voltage noise of the analog front-end should be therefore less than $2 \,\text{nV}/\sqrt{\text{Hz}}$.

4.6.2 Design and Implementation

The complete circuit diagram and PCB layout of the analog front-end are available in Appendix B.



Fig. 4.29 The circuit schematic of the V/I converter based on the op-amp differential amplifier

Low-Noise Amp

To realize a differential amplifier with such a small input bias current, N-channel J-FET is generally used for the input of the first stage. Figure 4.28 shows the circuit schematic of typical FET differential amplifier. For DC amplifier, two FET should be thermally coupled to suppress temperature drift and distortion. Toshiba 2SK389, dual J-FET monolithic N-channel, is historically used for this type of application, although it has been discontinued. Liner Integrated Systems, however, now provides LSK389, which is a drop-in replacement for 2SK389, so we adopted it in this design. The LSK389 (and 2SK389) has a moderate input capacitance of 25 pF, but it can be much larger due to the mirror effect when the gain at this stage is large. We therefore made the gain to be 6 dB.

For the second stage, we adopted a 40 dB op-amp differential amplifier using the AD8099 by Analog Devices Inc. Thanks to its small noise level of $0.95 \,\mathrm{nV}/\sqrt{\mathrm{Hz}}$, the second stage does not add more noise even though the gain of the first stage is small.

V/I Converter

Figure 4.29 shows the circuit schematic of typical op-amp V/I converter. When $R_1 = R_2 = R_3 = R_4$, the current through the load resistor, R_L , is only determined by V and R and given by I = V/R regardless to the resistance value at R_L . When R_L is zero, the op-amp does not amplify the input. When R_L is non-zero, the op-amp amplifies the input so that the output voltage of the op-amp, V_{out} , becomes

$$V_{\rm out} = \frac{R + R_{\rm L}}{R} V. \tag{4.58}$$

Therefore, the op-amp gain in the circuit is given by $(R + R_L)/R$, and it limits the maximum load resistance.

In our case, the resistance R can be as small as 50 Ω , while the typical wire resistance is ~ 200 Ω , which makes the op-amp gain to be 5. The maximum input to the op-amp is 1 V_{pp}, and the maximum output is therefore 5 V_{pp} when the gain is 5. As a result, the ±5 V power supply is sufficient.

The output current resolution is generally determined by the input bias current of the op-amp used as a voltage-follower. We selected the Analog Devices' ADA4817-2, which is dual unity-gain stable op-amp with FET inputs. The typical input bias current is 2 pA, which is small enough for our application.



Fig. 4.30 The measured gain (left) and noise (right) of the analog front-end

Other Features

To suppress the low-frequency signal variation in the SQUID output, the DC FLL circuit is also implemented. For the integrator we adopted the famous Texas Instruments' OPA627, which has very small bias current and voltage offset and is well-suited for a precision integrator. The time constant is set to $\sim 150 \,\mu s$.

The current source for the SQUID bias is also implemented. It is just a $51 \text{ k}\Omega$ resistor and a voltage source using DAC.

To control the parameters of DC FLL circuit, the Atmel AVR microcontroller, AT90USB162, is also on board. The parameters can be adjusted from a host computer connected to the front-end circuit by an USB cable. To suppress the noise from the clock, the microcontroller falls into the sleep mode and stops the clock when the USB cable is disconnected. Moreover, a dual channel slow speed ADC is also on board to read the amplified SQUID output for obtaining $\Phi-V$ characteristics when adjusting the operating point.

PCB Layout

The PCB was designed for the four-layered board. Figure C.2 in Appendix C shows the PCB layout of the component side and the solder side. The inner two layers are for +5 V and -5 V.

4.6.3 Measurements

Figure 4.30 shows the measured gain and noise of the developed analog front-end. The gain is almost 40 dB, and the cut off frequency is > 20 MHz. The low-cut off is due to the AC-coupled output. The noise is $< 2 \text{ nV}/\sqrt{\text{Hz}}$ for f > 1 kHz, and shows the flat characteristic for f > 1 MHz.

Chapter 5

Experiment of TES Readout

5.1 Objective

As a demonstration experiment of the developed readout system, we have performed an end-to-end test using a real TES array and a X-ray source. The primary objective of this experiment is to see if the developed system is able to 1) bias the TES to its working point, 2) operate the SQUID in FLL, and 3) collect X-ray pulses.

5.2 Setups

5.2.1 Refrigerator

For a refrigerator, we used the liquid-helium-free ³He-⁴He dilution refrigerator developed by Taiyo Nippon Sanso Corporation (Figure 5.1), which was originally designed for the microcalorimeter spectrometer system mounted on a transmission electron microscope [35, 12, 20]. It has a unique feature that the Gifford-McMahon (GM) cycle cryocooler and the dilution refrigerator are separated to prevent the vibration of GM cryocooler from being transmitted to the cryogenic stage.



GM Cryocooler

Fig. 5.1 The dry ³He-⁴He dilution refridgerator



Fig. 5.2 The cryostage setup



Fig. 5.3 The equivalent circuit for the cryostage setup

Table 5.1	LC	filter	setu	\mathbf{ps}
-----------	----	--------	------	---------------

Channel	TES Address	Inductance	Capacitance	Murata Part#	Frequency
		(nH)	(nF)		(MHz)
1	83	470	8.2	GRM2195C1H822JA01D	2.5
2	73		6.8	GRM2195C1H682JA01D	2.8
3	81		5.6	GRM2195C1H562JA01D	3.1
4	72		4.7	GRM2195C1H472JA01D	3.4

5.2.2 Cryogenic Stage

Figure 5.2 shows the cryostage setup and Figure 5.3 shows its equivalent circuit.

TES

The TES used in this experiment is the Ti/Au-bilayer (40/90 nm) 8×8 array developed by our research group (Figure 5.4). The chip size is 5.2×5.2 mm. The size of each TES pixel is $180 \times 180 \,\mu$ m, and the Au absorber with 600 nm in thickness and $120 \times 120 \,\mu$ m in size is EB-evaporated at the center.

SQUID and LC Filter

Although we developed the 4-input low-power SQUID with built-in bandpass filters in Chapter 3, we used the 4-input ISAS-G15 SQUID with inductors, and attached capacitors externally, since the resonance frequencies



Fig. 5.4 8×8 TES array

of the bandpass filter are higher than designed values. The capacitors are Murata's surface mount (2012/0805) multilayer ceramic capacitors with COG dielectric. The used capacitors and the designed resonance frequencies are summarized in Table 5.1. The PCB patterns of the TES–SQUID–Capacitor loop are solder-plated for a smaller residual resistance.

X-Ray Source

We used a ⁵⁵Fe X-ray radiation source. It emits Mn K α and Mn K β X-rays. The K α shows a doublet structure, K α_1 and K α_2 , with energies of 5.89875 keV and 5.88765 keV. The energy of the K β is 6.486 keV. The emission ratio of those lines are K α_1 : K α_2 : K $\beta = 20$: 10 : 3.

The source is screw-mounted on top of the TES (Figure 5.5 left).

Mounting to the Refrigerator

We first mounted the cryostage to the coldhead of the refrigerator, then mounted the Cryoperm magnetic shielding to shield the entire cryostage from the external magnetic field (Figure 5.5). The electrical wires used in the refrigerator are Nb-Ti twisted-pair woven looms, which is relayed with constantan twisted-pair woven looms halfway through the room temperature connectors.

5.2.3 Room-Temperature Front-End

The developed analog front-end was directly mounted to the refrigerator port (Figure 5.6). It is sealed in an aluminum diecast enclosure as a Faraday cage. The front-end and the digital system are connected using short ($\sim 30 \text{ cm}$) coax cables. The USB cable is connected to the front-end when we adjust the SQUID to the operating point, but it is disconnected during the measurement to suppress noises from the clock of the microcontroller.

5.2.4 SQUID Setup

The SQUID bias current, I_{SB} , was set to 17 μ A during the measurements. Figure 5.7 shows the Φ -V curve at 200 mK and the operating point of the SQUID. The DC feedback was disabled all the time.



Fig. 5.5 Mouting to the refrigerator: the cryostage mounted to the coldhead of the refrigerator (left) and the cryostage covered by the Cryoperm magnetic shielding (right)



Fig. 5.6 The analog front-end directly mounted to the refrigerator port



Fig. 5.7 The Φ -V curve ($I_{\rm SB} = 17 \,\mu A$) at 200 mK and the operating point during the measurements



Fig. 5.8 The LC filter characteristics at 150 mK (TES super) and 240 mK (TES normal)

5.3 Results

5.3.1 LC Filter Characteristics

We first measured the LC filter characteristics. The SQUID was used in the open-loop, thus a TES bias current was kept low so that the SQUID output at resonance frequencies is not folded. The frequency of the TES bias current was then swept from 2.3 to 3.6 MHz in 1 kHz steps, and the SQUID output was measured.

Figure 5.8 shows the measured LC filter characteristics at 150 mK (TES super) and 240 mK (TES normal). At 150 mK, we see four resonance frequencies at 2.5, 2.8, 3.0, and 3.3 MHz, although the filter transmissivities for 2.5 and 3.0 MHz are very small compared to that for 2.8 and 3.3 MHz, and the resonances are almost invisible for 2.5 and 3.0 MHz at 240 mK. We have seen smaller transmissivities for particular frequencies before, but it was the first time to see such small transmissivities.

We extracted the resonance frequencies, which the SQUID output becomes the maximum, from the characteristic at 150 mK and summarized in Table 5.2. The resonance frequencies, however, become slightly larger when

	Channel	Resonance Frequency (open)) Resonance Frequency (FLL)	
		(MHz)	(MHz)	
	1	2.502	2.505	
	2	2.763	2.767	
	3	3.035	3.038	
	4	3.288	3.290	
10^3 10^3 10^1 10^2 10^1 10^1 10^2		(^{surv}) • Ch. 1 (2.505 MHz) • Ch. 2 (2.767 MHz) • Ch. 3 (3.038 MHz) • Ch. 4 (3.290 MHz)	10 ⁻⁵ 10 ⁻⁶ ••• Ch. 1 (2.505 MHz) ••• Ch. 2 (2.767 MHz) ••• Ch. 3 (3.038 MHz) ••• Ch. 4 (3.290 MHz)	
140 160) 180 200	220 240 260 280	100140160180200220240260	280
	Tempera	ature (mK)	Temperature (mK)	

Table 5.2 The measured resonance frequencies of the LC filter

Fig. 5.9 The measured R-T characteristics (left) and the TES current during the R-T measurement (right)

in FLL due to the nulled self inductance of the SQUID input coil. The resonance frequencies in FLL are also measured, and summarized in the same table.

5.3.2 *R*–*T* Measurements

Figure 5.9 (left) shows the measured R-T characteristics of each TES channel for temperature from 150 to 270 mK in 5–10 mK steps. During the measurement, the TES bias currents for each channel were fixed to values that the TES currents at 150 mK become $\leq 3 \,\mu A_{\rm rms}$ to prevent TES from self-heating (Figure 5.9 right). When the SQUID is in FLL, the TES resistance $R_{\rm TES}$ is given by

$$R_{\rm TES} = \left(\frac{I_{\rm TB}}{I_{\rm TES}} - 1\right) R_{\rm S} \tag{5.1}$$

$$= \left(\frac{V_{\rm TB}}{V_{\rm FB}}\frac{R_{\rm FB}}{R_{\rm TB}}\frac{M_{\rm in}}{M_{\rm FB}} - 1\right)R_{\rm S}$$
(5.2)

$$= \left(\frac{TB}{FB}\frac{R_{\rm FB}}{R_{\rm TB}}\frac{M_{\rm in}}{M_{\rm FB}} - 1\right)R_{\rm S},\tag{5.3}$$

where I_{TB} is the TES bias current, I_{TES} is the TES current, R_{S} is the TES shunt resistance, which is $5 \text{ m}\Omega$, V_{TB} is the TES bias voltage, V_{FB} is the feedback voltage, R_{FB} is the feedback resistance, R_{TB} is the TES bias resistance, M_{in} is the mutual inductance of the SQUID input coil, which is 83.26 pH, M_{FB} is the mutual inductance of the feedback coil, which is 76.62 pH, TB is the normalized TES bias value in the digital system, and FB is the normalized feedback value in the digital system. The TES resistances in Figure 5.9 (left) were calculated using the conversion equation above. The TES current, I_{TES} , is given by

$$I_{\rm TES} = \frac{FB}{R_{\rm FB}} \frac{M_{\rm FB}}{M_{\rm in}} \frac{1}{N},\tag{5.4}$$



Fig. 5.10 The TES shunt resistor adjusted R-T (left) and the TES bias level adjusted R-T (right)

Channel	Frequency	Correction Factor
	(MHz)	
1	2.505	0.088
2	2.767	0.614
3	3.038	0.139
4	3.290	1.147

Table 5.	3 Т	The	obtained	correction	factors	for	each	channe

where N is the output scaler of the digital system. In R-T measurements, R_{TB} , R_{FB} , and N were set to 250.5 Ω , 2.511 k Ω , and 2, respectively.

In the R-T result, we see huge inconsistencies in resistances among channels, even though the feedback currents are consistent to each other. It indicates that either (or both) the impedance of TES shunt resistor or (and) the impedance of the TES bias is (are) not frequency-independent. Some of TES in this array have been measured in a DC setup, and they showed consistent normal resistances, ~ 60 m Ω . Therefore we adjusted the R-T by changing 1) R_S , the TES shunt resistance value, and 2) TB, the TES bias value, so that the TES resistance at 270 mK becomes 60 m Ω . Figure 5.10 shows the R-T that the TES shunt resistor is adjusted for each channel (left) and the TES bias level is adjusted for each channel (right). While they show good agreement in resistance, the TES bias adjusted R-T shows the better result. Therefore, we define an adjusted TES bias level, TB', as

$$TB' = \gamma TB,\tag{5.5}$$

where γ is a correction factor. The correction factors for each channel in this experiment are summarized in Table 5.3, and we will use adjusted TES bias levels hereafter.

5.3.3 Non-multiplexing *I*–*V* Measurements

We first individually measured I-V characteristics for each TES channel. The output scaler of the digital system is set to 1. For channel 2 and 4, we measured the I-V at bath temperature 150, 175, 190 and 205 (Ch. 2 only) mK. For channel 1 and 3, we could not prevent TES from being super even though the maximum TES bias voltage was applied. Consequently, we changed the TES bias resistance, R_{TB} , from 250.5 Ω to 75.6 Ω when measuring channel 1 and 3, and succeeded to measure the I-V, but only at temperature 190 mK. Figure 5.11 shows the measured I-V characteristics.



Fig. 5.11 The measured I-V characteristics of each channel in non-multiplexing

5.3.4 Non-multiplexing Pulse Collections

At first, we collected X-ray pulses in channel 2 at 150 mK for various TES bias currents. Figure 5.12 shows the averaged pulses for 2-ms length 1,000 pulses for each TES bias current. As the TES bias current decreases, the pulse height becomes larger, and it starts oscillating. It was actually happening in all the channels. The oscillation is due to the instability in the ETF. According to Irwin and Hilton [16], the TES under the ETF is stable when

$$\tau > (\mathcal{L}_0 - 1)\tau_{\rm el},\tag{5.6}$$

where τ is the TES time constant, \mathcal{L}_0 is the TES loopgain at DC, and τ_{el} is the electrical time constant given by

$$\tau_{\rm el} = \frac{L}{R_L + R_0 (1+\beta)},\tag{5.7}$$

where L is the inductance in the TES–SQUID loop, R_L is the resistance in the loop, R_0 is the TES resistance, and β is the TES current sensitivity, which is defined as

$$\beta \equiv \frac{d\log R}{d\log I} = \frac{I}{R} \frac{dR}{dI}.$$
(5.8)



Fig. 5.12 The averaged pulses in channel 2 at 150 mK for various TES bias currents

In our configuration, L is the double of the inductance in the filter, $\sim 1 \,\mu\text{H}$, R_L is the shunt resistance, $5 \,\mathrm{m}\Omega$, and R_0 depends on the TES bias current, but we here assume the typical value, $50 \,\mathrm{m}\Omega$. \mathcal{L}_0 also depends on the TES bias current, and is typically < 50 for our TES. Ignoring β for simplicity, we now obtain

$$\tau \gtrsim 1 \times 10^{-3} \left(\frac{\mathcal{L}_0}{50}\right) \left(\frac{L}{1\,\mu\text{H}}\right) \left(\frac{5\,\mathrm{m}\Omega + 50\,\mathrm{m}\Omega}{R_L + R_0}\right) \quad (\text{s}).$$
(5.9)

Therefore, if the TES time constant is larger than 1 ms, the TES is stable. Unfortunately, the absorber of the TES is small and thin, which makes the heat capacity to be $\sim 0.2 \text{ pJ/K}$, while the thermal conductivity of the TES is $\sim 2.5 \text{ nW/K}$. The time constant of the TES is therefore $\sim 80 \,\mu\text{s}$, which is smaller than the required time constant by more than an order of magnitude.

Once oscillation happens, a good energy resolution is never expected. Figure 5.13 shows the oscillating pulse waveforms in channel 4 (the channel number in the figure is dummy). As we can see in the zoomed figure at right, the peak positions do not match, which breaks the similitude of pulses that is assumed in the optimal filtering.

As a result, the obtained energy resolutions are very poor. They are summarized in Table 5.4. The best energy resolution is 48 eV FWHM @ 5.9 keV at channel 1. There is a trend that at first the energy resolution becomes better as the TES bias current decreases, but it suddenly becomes worse as the TES starts oscillating.



Fig. 5.13 The oscillation due to the unstable ETF of TES: the oscillating pulse waveforms (left) and their zoomed peaks (right)

Channel	Thath	TES Bias	SNR	Energy Resolution		
	bath			Baseline @ 5.9 keV		
	(mK)	$(\mu A_{\rm rms})$	(dB)	(eV FWHM)	(eV FWHM)	
1	190	241	49	42	140	
		231	51	38	48	
		221	52	32	73	
		216	52	25	69	
2	150	529	41	110	134	
		518	44	63	116	
		508	46	51	128	
		497	48	49	152	
	175	444	43	76	130	
		434	46	52	98	
		423	46	59	122	
	190	360	46	51	100	
		349	48	38	108	
		339	48	100	102	
3	190	325	45	60	134	
		317	47	44	126	
		309	49	163	218	
4	150	553	38	146	244	
		534	43	71	118	
		514	46	44	174	
	175	474	36	208	216	
		455	42	90	164	
		435	44	64	153	
	190	375	41	95	134	
		356	45	61	136	
		340	47	54	140	

Table 5.4 The obtained energy resolutions in non-multiplexing pulse collections



Fig. 5.14 The simultaneously measured I-V characteristics multiplexing channel 2 and 4

Channel	$T_{\rm bath}$	TES Bias	SNR	Energy Resolution		
				Baseline	$@ 5.9 \mathrm{keV}$	
	(mK)	(μA_{rms})	(dB)	(eV FWHM)	(eV FWHM)	
2	175	433	47	51	130	
4		455	42	85	130	
2	190	349	49	35	116	
4		356	46	48	140	

Table 5.5 The obtained energy resolutions in multiplexing pulse collections for channel 2 and 4

5.3.5 Simultaneous *I–V* Measurements

Next we simultaneously measured I-V multiplexing channel 2 and 4. The TES bias resistance is back to 250.5Ω , and the output scaler is set to 1/2. Figure 5.14 shows the measured I-V. Since we set the output scaler to 1/2, the maximum TES bias is reduced by half, and the measurable range was limited by that. They are still in agreement with the non-multiplexing result.

We also succeeded to measure I-V simultaneously multiplexing all channels only at 205 mK. The TES bias resistance and the feedback resistance needed to be changed to 55.0 Ω and 1258 Ω , respectively. Figure 5.15 shows the measured I-V.

5.3.6 Simultaneous Pulse Collections

We first simultaneously collected pulses in channel 2 and 4 at 175 mK and 190 mK, and then in all the channels at 205 mK.

For simultaneous pulse collection for channel 2 and 4, R_{TB} , R_{FB} , and N were set to 250.5 Ω , 2511 Ω , and 1/2, respectively. Table 5.5 summarizes the obtained energy resolution. They are still suffered by the oscillation.

For simultaneous pulse collection for all the channels, R_{TB} , R_{FB} , and N were set to 55.0 Ω , 1258 Ω , and 1/2, respectively. Table 5.6 summarizes the obtained energy resolution. While the energy resolutions are all over 100 eV, we succeeded in the 4-channel simultaneous pulse collection in FDM from the single-staged cryogenic setup for the first time.



Fig. 5.15 The simultaneously measured I-V characteristics multiplexing all channels

Table 5.6 The obtained energy resolutions in multiplexing pulse collections for all the channels

Channel	$T_{\rm bath}$	TES Bias	SNR	Energy Resolution		
				Baseline	$@ 5.9 \mathrm{keV}$	
	(mK)	$(\mu A_{\rm rms})$	(dB)	(eV FWHM)	(eV FWHM)	
1	205	150	48	58	146	
2		207	45	65	206	
3		212	40	128	252	
4		230	40	118	212	

Chapter 6

Summary

6.1 Summary

In this thesis, we have freshly developed the entire FDM readout system for large-format TES microcalorimeter arrays toward the future space missions. The developed system consists of the low-power multi-input SQUID with built-in bandpass filters, the high-frequency digital FLL electronics, and the analog front-end optimized for bridging the low-power SQUID and the digital electronics. The cold electronics are optimized for the simple single-staged configuration, and the warm electronics are optimized for the high-density signal multiplexing to support the simple cryogenic setup.

The unique feature of the developed SQUID is its very low heat dissipation, $\sim 20 \text{ nW}$, while having the sufficient gain, > 100 V/A. It also has the very low noise characteristic, $< 10 \text{ pA/}\sqrt{\text{Hz}}$ @ 4K. With these features, the SQUID can suffice to be the only amplifier in the cold electronics even though it can be placed at the cryogenic stage below 100 mK. Using the developed low-power SQUID, we also developed the multi-input SQUID chip with built-in bandpass filters. Within the size of $2.5 \times 2.5 \text{ mm}$, it carries the low-power SQUID, the TES shunt resistor, and the LC bandpass filters for four channels. We also developed the same-size extension chip that consists of the bandpass filters for four other channels, and it can be attached to the multi-input SQUID chip with only two bonding wires. With these chips, the cryogenic stage setup is dramatically simplified, and the feasibility of the large-format TES array is demonstrated.

The digital FLL electronics consists of the fast FPGA and the fast ADC/DAC module. It has the large signal bandwidth, > 10 MHz, and can multiplex up to 16 channels without deteriorating the required SNR, > 60 dB. There is still some margin in the hardware resource, and the number of multiplexing channels may be upgraded up to 32 without adding more hardwares. The SNR is also expected to be kept > 60 dB even in 32-channel multiplexing. The maximum unity-gain bandwidth in FLL is 76.8 kHz while having the sufficient phase margin of $\sim 60^{\circ}$, which enables the stable FLL in the wide signal band. To realize the high-density multiplexing with the limited amount of system resource in satellites, the data rate needs to be greatly reduced without deteriorating the required SNR. In order to do that, waveforms are first triggered and extracted in the high sampling-rate at ~ 4 MHz, and then aligned and down-sampled to 240 ksps. With this method, the required data amount for a 2-ms waveform is reduced to only 1 kB, keeping the required system data transfer rate to a practical number even for thousands of signal multiplexing.

To bridge the low-power SQUID and the digital FLL electronics, the analog front-end is also developed. It mainly consists of the low-noise amp to amplify the SQUID output without adding extra noises, and the V/I converters to convert the voltage outputs of two DAC to the TES bias current and the feedback current, keeping the high output impedance. The low-noise amp has the sufficient gain, 46 dB (40 dB when 50 Ω -terminated), and the sufficient low noise characteristic, $< 2 \,\mathrm{nV}/\sqrt{\mathrm{Hz}}$ @ 10 kHz. The V/I converter has the wide signal bandwidth

over 20 MHz, and can supply up to $30 \,\mathrm{mA_{pp}}$ for the DAC maximum output of $1 \,\mathrm{V_{pp}}$, which makes possible to multiplex more than 10 channels.

Finally, we performed the TES readout experiment to demonstrate the developed readout system, and successfully multiplexed four TES for the first time. Although the energy resolution suffered from the unstable ETF due to the small heat capacity of the TES, we could readout X-ray pulses from four channels simultaneously.

The results demonstrate the feasibility of the large-format 256–400 TES arrays for the DIOS missions, as well as the possibility of the extra-large-format over-thousand TES arrays for future X-ray space missions.
A

CRAVITY

A.1 AIST Nb Standard Process (STD3)

All the SQUIDs we developed were fabricated in the Clean Room for Analog & Digital superconductiVITY (CRAVITY) at National Institute of Advanced Industrial Science and Technology (AIST), in accordance with the SRL Nb Standard Process (STD3). Figure A.1 shows the device structure of STD3. There are four conductive layers, GP, BAS, COU and CTL, but we only used three layers except GP. Table A.1 shows the detailed layer structure of STD3. The AU layer and the PL layer are optional. We used AU pads. The resistor layer uses molybdenum in standard, however, it can not be used below 1 K. We therefore used palladium-gold (Pd/Au) for the resistor layer.

The standard design rule in the STD3 follows:



Fig. A.1 The device structure of SRL Nb Standard Process (STD3) [23]

Layer	Function	Material	Thickness
			(nm)
GP	Ground	Nb	300
	Insulating layer	SiO_2	150
RES	Resistor	Mo	35
	Insulating layout	SiO_2	150
\mathbf{RC}	Contact between RES and BAS		
GC	Contact between GP and BAS		
BAS	Conductive layer	\mathbf{Nb}	300
$_{\rm JP}$	JJ protection	Al, AlOx	
JJ	Josephson junction	\mathbf{Nb}	150
	JJ protection (anodic oxidation)	Nb_2O_5	(20)
	Insulating layout	SiO_2	400
BC	Contact between BAS and COU		
JC	Contact between JJ and COU		
COU	Conductive layer	\mathbf{Nb}	400
	Insulating layout	SiO_2	500
$\mathbf{C}\mathbf{C}$	Contact between COU and CTL		
CTL	Conductive layer	\mathbf{Nb}	500
AU	Pad (spattering)	Au	300
\mathbf{PL}	Pad (plating)	Au	3000

Table A.1 The layer structure of SRL Nb Standard Process (STD3) [23]

• Minimum line width	$1.5\mu{ m m}$
• Minimum spacing	$1.0\mu{ m m}$
• Minimum junction size	$1.1\mu\mathrm{m}~(1.0\mu\mathrm{m} \text{ after shrink})$
\cdot Shrink size of junctions	$0.1\mu{\rm m}~(<0.3\mu{\rm mfor}$ junctions less than $1.0\mu{\rm m}^2)$
\cdot Critical current density of junctions	$10\mathrm{kA/cm^2}$
\cdot Sheet resistance for RES	2.4Ω
• Minimum contact size	$1.0\mu\mathrm{m}~(0.7\mu\mathrm{m}$ for JC)
• Alignment margin	$0.5\mu\mathrm{m}~(0.25\mu\mathrm{m}$ for some layers)
• Shrink size for BAS	$0.2\mu{ m m}$
\cdot Shrink size for COU and CTL	0
• Shrink size for RES	0

The critical current density of junctions in standard, 10 kA/cm^2 , is too large for our SQUIDs, so we used 1 kA/cm^2 .

Finally, alignment margins, minimum spacing between layers, minimum line width in the STD3 are summarized from Table A.2 to Table A.4, respectively.

	GP	RES	GC	RC	JJ	JP	BAS	JC	BC	COU	CC	CTL
GP		×	×	×	×	×	×	×				
RES	×		×	0.50	1.00							
GC	×	×		\times	×	0.25	0.50	×				
\mathbf{RC}	×	0.50	×		0.50	0.25	0.50					
JJ	×	1.00	×	0.50		0.50		0.25				
$_{\rm JP}$	×		0.25	0.25	0.50	• • •						
BAS	×		0.50	0.50			• • •		0.50			
JC	×		×		0.25				×	0.50		
BC							0.50	×		0.50		
COU								0.50	0.50	•••	0.50	
$\mathbf{C}\mathbf{C}$										0.50		0.50
CTL											0.50	

Table A.2 STD3 alignment margins (unit: μ m) [23]

 \times : No overlap.

Table A.3 STD3 minimum spacing between layers (unit: μ m) [23]

	GP	RES	GC	RC	JJ	JP	BAS	JC	BC	COU	CC	CTL
GP	1.0	1.0			1.0		0.45					
RES	1.0	1.0	1.0									
GC		1.0	1.0		1.0							
\mathbf{RC}				1.0								
JJ	1.0		1.0		1.0							
$_{\rm JP}$						1.0						
BAS	0.45						1.0					
JC								1.0				
BC									1.0			
COU										1.0		
$\mathbf{C}\mathbf{C}$											1.0	
CTL												1.0

Table A.4 STD3 minimum line width (unit: μ m) [23]

	GP	RES	GC	RC	JJ	JP	BAS	JC	BC	COU	CC	CTL
Minimum width	1.0	1.5	1.0	1.0	1.1	1.5	1.5	0.7	1.0	1.5	1.0	1.5

В

Hardware Registers and Addresses for Digital Electronics

						$\left - \right $																			Channel Stats (Ch 15) Channel Stats (Ch 16)	78 7C	1.1	- 0				00
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				_		+	+		+	+		+	+			+			+			_			Channel Stats (Ch 10)	£			0	-	-	0
				_		+	+		+	-			-	1		+	-		-	_		_			Channel Stats (Ch 9)	8	r.	•	0		<u> </u>	0
						+	_		+	+			-	1		+			+			_			Channel Stats (Ch 8)	50		o	.	•	.	0 0
				_		+	+		+	-		+	+	1		+			+	_		-			Channel Stats (Ch 7)	58		•	-	•	-	0
							_			_				_		_						_			Channel Stats (Ch 6)	5 4	•		0	0	-	0
																									Channel Stats (Ch 5)	50		•	0	0	-	0
																									Channel Stats (Ch 4)	4 0	i.		-	0		0
																									Channel Stats (Ch 3)	48	1	•	-	0		0
																									Channel Stats (Ch 2)	44	i.		0	0	-	0
* * Unsigned	*	*	*	*	*	*	*	*	*	*													RO	Triggered Count								
											*	*	*	*	*	*	*	*	*	*	*		RO	Reserved								
0: Idle 1: Running																						*	RO	Idle	Channel Stats (Ch 1)	40		0	0	0		0
																										ЗC	•				0	0
																										38	•	•	-		0	0
																										34	•		0		0	0
																										8	1	•	0		0	0
																										2C	1		-	-	0	0
																										28	•	•	-	-	0	0
																										24	1		0	-	0	0
																										20	1	•	0	-	0	0
																									Reserved	10	•		-	0	0	0
* * Unsigned	*	*	*	*	*	*	*	*	*	*													RW	Next Trigger (15:0)								
Unsigned											*	*	*	٠		*	٠	*	*	*		*	RW	Previous Trigger (15:0)	Trigger Threshold	18		•		•	0	0
Unsigned																									Reserved	14	•		0	0	0	0
 * Unsigned / Big-endian 	*	*	*	*	*	*	*	*	*	*	*		*	*	*	*	*	*	*	*	*	*	RW	Data	Modulation Data	10	•	•	0	0	0	0
 * Unsigned / Big-endian 	•	*	•	*	•	*	*	*	*	*	*		*	*		*	*	*	*	*	*	•	RO	Data	Data FIFO	8	•		-	0	0	0
* * Unsigned	*	*	*	*	*	*	*	*	*	*													RO	FIFO Counts								
											*		*	*		*	*	*	*	*	*	•	RO	Reserved	Device Stats	8	•	•	-	0	0	0
*	*	*	*	*	*	*	*																	Reserved								
Unsigned								*	*	*													RW	Mod Write Ch (3:0)								
Unsigned											*		*	*		*							RW	Trigger Delay (7:0)								
0: 256 1: 512 2: 1024 3: 2048																	*	*					RW	Record Length (1:0)								
0:2 1:4 2:8 3:16																			*				RW	Output Scaler (1:0)								
0: Disabled 1: Enabled							-		F	-						-				*		-	RW	No Output Scaling								
0: Disabled 1: Enabled																					*		RW	Loopback								
0: Disabled 1: Enabled							_		-	-		-	-			-							RW	IRQ Enabled								
1: Reset																						*	WO	Reset	Device Ctrl	24			0	0	0	0
* * Unsigned	*																						RO	Number of Channels								
Unsigned		*	*	*	*	*	*	*	*	*													RO	Revision								
0xBBFB											*	*	*	*	*	*	*	*	*	*	*	*	RO	Signature	Version	8		0	0	0	0	0
1 0	N	4	сл	7 6	8	6 0	11 1	12	4 13	15 14	16	3 17	19 14	20 .	2 21	23 22	24 2	5 25	27 26	28	29	31 30				Hex	0	2 1	ω	CT N	σ	7
Description										ß	Dat												R/W	Signal Name	Register Name			ess	Addr			
											1																					



27 36 25 24 23 22 21 20 19 19 15 15 15 15 11 10 9 8 7 6 5 4 3 2 1 0		0: Ready 1: Clear	0: Disabled 1: Enabled	0: Disabled 1: Enabled	O: Level 1: Count	O: Frequency 1: Phase		O: Disbled O: Disbled		· · ·	See Trigger Mode	the second se	Signed (4 to +3)		Unsigned	Public de la construction de la																														
V 31 30	· ·	*	>	>	>	>	>			>	>	>	>	>	•	>																														
SR S	a	S.	ž	Å	ž	Å	2	S.	2	ž	N.	ž	S.	Å	NR (:0) RM																			_	_										
Signal Name	Enable	Clear	Feedback En	Integrator En	Trigger Level/Count Sel	Frequency/Phase Sel	Baseband Gain (1:0)	Modulation En	Reserved	Derivation Mode (1:0)	Trigger Mode (3:0)	Reverse	Gain (2:0)	Amplitude (11:0)	Frequency/Phase (15:0)	Trigger Level/Count (15																														
Register Name	Channel Ctrl (Ch 1)																Channel Ctrl (Ch 2)		Channel Ctrl (Ch 3)		Channel Ctrl (Ch 4)		Channel Ctrl (Ch 5)		Channel Ctrl (Ch 6)		Channel Ctrl (Ch 7)		Channel Ctrl (Ch 8)		Channel Ctrl (Ch 9)		Channel Ctrl (Ch 10)		Channel Ctrl (Ch 11)		Channel Ctrl (Ch 12)		Channel Ctrl (Ch 13)		Channel Ctrl (Ch 14)		Channel Ctrl (Ch 15)		Channel Ctrl (Ch 16)	
7 6 5 4 3 2 1 0 Hex															1 0 0 0 0 1 - 84		1 0 0 1 0 - 88	1 0 0 1 1 8C	1 0 0 1 0 0 90	1 0 0 1 0 1 94	1 0 0 1 1 0 98	1 0 0 1 1 1 9C	1 0 1 0 0 0 A0	1 0 1 0 0 1 - A4	1 0 1 0 1 0 A8	1 0 1 0 1 1 AC	1 0 1 1 0 0 - B0	1 0 1 1 0 1 B4	1 0 1 1 1 0 - B8	1 0 1 1 1 1 BC	1 1 0 0 0 · · C0	1 1 0 0 0 1 C4	1 1 0 0 1 0 - C8	1 1 0 0 1 1 CC	1 1 0 1 0 0 D0	1 1 0 1 0 1 D4	1 1 0 1 1 0 D8	1 1 0 1 1 1 DC	1 1 1 0 0 0 E0	1 1 1 0 0 1 - E4	1 1 1 0 1 0 - E8	1 1 1 0 1 1 EC	1 1 1 1 0 0 F0	1 1 1 1 0 1 - F4	1 1 1 1 1 0 - F8	

Fig. B.2 The hardware registers and addresses for configuration parameters of the digital electronics (part 2 of 2)

С

Design of Analog Front-End



Fig. C.1 The circuit schematic of BBFB analog front-end



Fig. C.2 Full-scale PCB layouts of BBFB analog front-end: the component side (left) and the solder side (right)

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