

# Energy-Efficient Analog-to-Digital Conversion for Ultra-Wideband Radio

by

Brian P. Ginsburg

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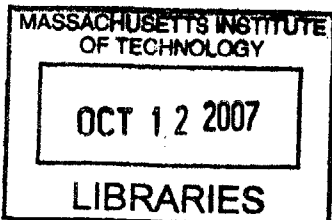
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## Abstract

In energy constrained signal processing and communication systems, a focus on the analog or digital circuits in isolation cannot achieve the minimum power consumption. Furthermore, in advanced technologies with significant variation, yield is traditionally achieved only through conservative design and a sacrifice of energy efficiency. In this thesis, these limitations are addressed with both a comprehensive mixed-signal design methodology and new circuits and architectures, as presented in the context of an analog-to-digital converter (ADC) for ultra-wideband (UWB) radio.

UWB is an emerging technology capable of high-data-rate wireless communication and precise locationing, and it requires high-speed ( $>500\text{MS/s}$ ), low-resolution ADCs. The successive approximation register (SAR) topology exhibits significantly reduced complexity compared to the traditional flash architecture. Three time-interleaved SAR ADCs have been implemented. At the mixed-signal optimum energy point, parallelism and reduced voltage supplies provide more than  $3\times$  energy savings. Custom control logic, a new capacitive DAC, and a hierarchical sampling network enable the high-speed operation. Finally, only a small amount of redundancy, with negligible power penalty, dramatically improves the yield of the highly parallel ADC in deep sub-micron CMOS.

Thesis Supervisor: Anantha P. Chandrakasan

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It has been my pleasure to work with the great students in Anantha's group and throughout MTL and the MIT circuits community. As a tribute to the high-data-rate UWB team, initially consisting of Dave, Fred, Johnna, and Raúl, and later Vivienne, Kyle, Nathan, and Mike, here is my poem describing our combined efforts, initially presented at the 2006 MTL Annual Research Conference:

*Once there was only narrowband radio  
It uses lots of power and goes kind of slow  
Along came UWB with speed and positioning for free  
And soon it will control the dataflow  
If you come to our poster, I trust you will find  
Our group's research will truly blow your mind  
A discrete RF prototype that has two horn antennae  
And a digital backend implemented in an FPGA  
With this flexible platform we were able to show  
Wireless transmission of Nathan's dancing video  
From this experience knowledge we did take  
And a custom chipset we did make  
With built-in pulse shaping sidebands keep away  
Also, it has a WLAN notch filter and an unmatched LNA  
Two parallel SAR ADCs deserve some looks  
And a baseband processor that reconfigures with many hooks  
Throughout our research we have made much progress  
To put these chips together at 100Megabps  
Now we ask, if it's not too bold  
Whether the baseband can operate in the sub-threshold  
In the future our challenge will be  
A new transceiver running off the power of a small battery  
And while I do not want to be a big booster  
To learn more you must come see our awesome poster*

I would especially like to thank Naveen for the many fruitful discussions along the way. For a while, we were the only two people at MIT pushing for the resurgence of the SAR architecture, and he pointed me in the correct direction many times, from Promitzer's self-timed bit-cycling to the hierarchical top-plate sampling network. He independently formulated the latter as a solution to my sampling problem, and I only afterwards discovered that it (along with many great ideas) had been previous published.

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In memoriam, Douglas C. Baker, 11/25/1980–10/1/2006.





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# Chapter 1

## Introduction

In February 2002, the Federal Communications Commission (FCC) approved the use of digital ultra-wideband (UWB) communication in the 3.1–10.6 GHz band [1]. UWB signals are defined by the FCC as having a 10 dB bandwidth that either exceeds 500 MHz or is greater than 20% of the center frequency (for center frequencies within 3.1–10.6 GHz, only the former constraint is meaningful, as any transmission that satisfies the latter must have a bandwidth greater than 500 MHz). UWB is a technology that is overlaid on spectrum allocated to other services, and the FCC therefore limits the average transmitted power density to less than  $-41.3$  dBm/MHz, making only short distance communication feasible. To utilize this spectrum for high-data-rate communications, the IEEE formed the 802.15.3a Task Group, which engaged in an ultimately futile effort to produce a single standard. By early 2004, the 802.15.3a Task Group had narrowed down its selection to two competing proposals.

The Multi-Band OFDM Alliance (MBOA), now part of the WiMedia Alliance, proposed transmitting the data using orthogonal frequency division multiplexing (OFDM) symbols [2], the same technique already commonplace in 802.11g and 802.11a devices. Data is transmitted in 528 MHz wide symbols consisting of 122 evenly spaced subcarriers, 100 of which carry the data. Consecutive OFDM symbols are modulated to one of two or three center frequencies, dependent on the mode, with only 9.47 ns to switch channels. MB-OFDM supports data

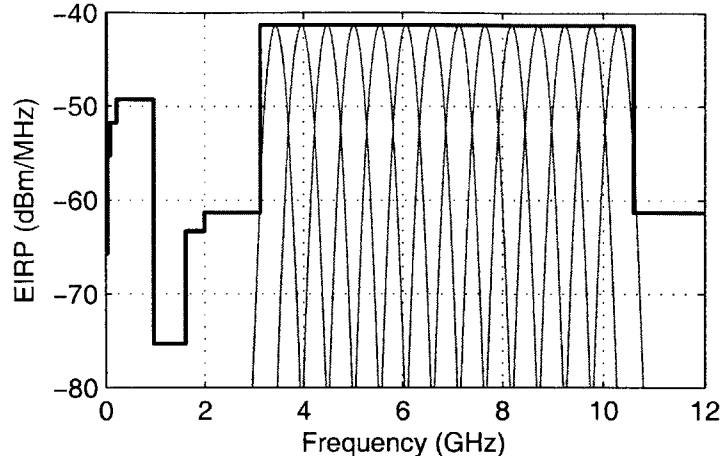


Figure 1-1: FCC UWB spectral mask and MIT transceiver frequency plan.

rates up to 480 Mb/s.

The UWB Forum proposed Direct Sequence UWB (DS-UWB), which transmits the data as closely spaced, narrow pulses that are modulated using binary phase shift keying (BPSK) or quaternary bi-orthogonal keying (4-BOK) [3]. These pulses are transmitted in one of two frequency bands (3.25 – 4.6 GHz or 6.6 – 9.3 GHz), with different spreading codes to further subdivide these two bands. The maximum supported data rate is 1.32 Gb/s.

In the end, the 802.15.3a Task Group deadlocked and disbanded in January without promulgating a standard. The WiMedia Alliance and UWB Forum both pledged to bring high-data-rate UWB devices to market.

The MIT Ultra-Wideband Project uses its own, pulse-based UWB architecture [4, 5]. 500 MHz wide Gaussian pulses are generated at baseband and upconverted to one of 14 channels in the 3.1–10.6 GHz band, shown fitting under the FCC spectral mask in Fig. 1-1. While the center frequencies chosen are identical to the MB-OFDM plan, frequency hopping is not employed. The pulses themselves are transmitted at a pulse repetition frequency (PRF) of 100 MHz, with each pulse BPSK-modulated, for a maximum data rate of 100 Mb/s.

Due to the strict radiated power limits, all of the proposed UWB systems use modulation schemes that can operate in low-signal-to-noise ratio (SNR) environments (e.g., QPSK/BPSK as opposed to 256-QAM). The large data rates are achieved through the wide



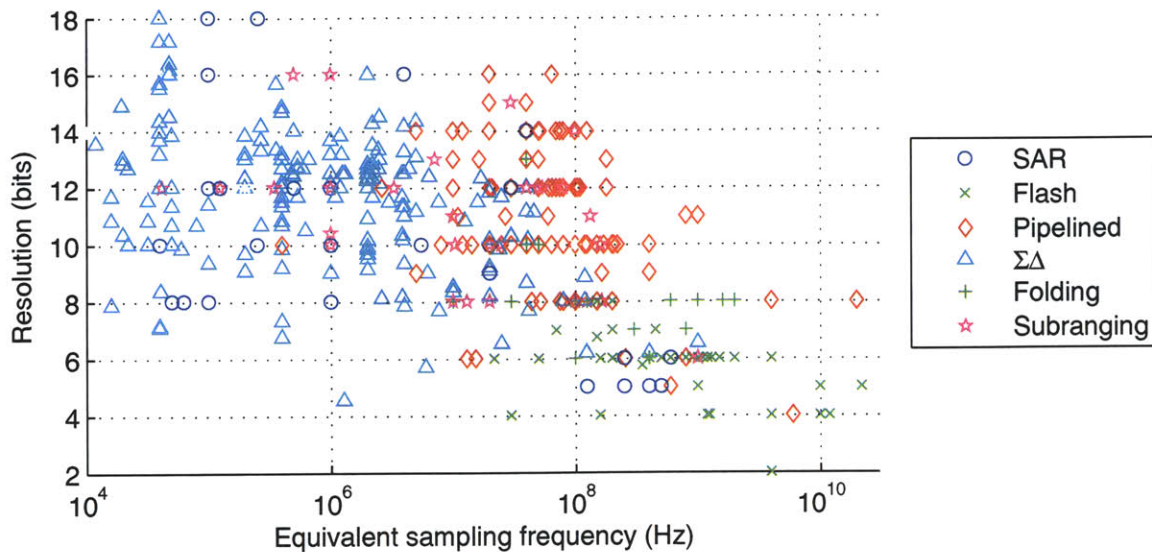


Figure 1-2: Scatter plot showing the distribution of the major ADC architectures across resolution and sampling rate. For oversampling converters, the resolution is plotted against twice the signal bandwidth of the converter.

other architectures.

Pipelined ADCs are used for high-speed, medium-resolution applications [23, 24]. They can achieve one conversion per clock period throughput and only a linear scaling in complexity with resolution; however, they rely on operational amplifiers at the heart of the multiplying digital-to-analog converter (MDAC) in each pipelined stage. Because it must be closed loop stable, this amplifier typically uses one or two high gain stages. Unfortunately, in deep sub-micron CMOS, the achievable gain per stage is limited because short-channel effects lower  $g_m r_o$  for a single transistor, and reduced voltage supplies restrict circuit techniques such as cascoding. Recent work has replaced the operational amplifier with open loop amplifiers [25] or current sources and comparators [26, 27], but this thesis explores other architectures with fewer scaling challenges than pipelined converters.

The charge redistribution SAR architecture was created in the 1970s at Berkeley [28, 29] and has been used extensively since then. A conventional SAR ADC consists of a digital-to-analog converter (DAC) driving a comparator, whose output is processed by control logic (called the SAR itself) that in turn drives the capacitor array. The SAR algorithm is a

bandwidth, which should lead to better transceiver energy efficiencies compared to a system that simply scaled the bandwidth of existing narrowband (e.g., 802.11g/a) systems [6]. Thus, even though high-speed analog-to-digital converters (ADCs) are required (500 MS/s for the MIT system), only 4 to 5 bits of resolution are needed for proper reception of both pulses and OFDM symbols [7–9].

The newly allocated spectrum is not being exclusively considered for high-data-rate applications. Several low-data-rate systems (up to around 20 Mb/s) with longer range and/or lower power consumption have been proposed, and a low-data-rate solution for personal area networks (PANs) has been standardized through the IEEE 802.15.4a Task Group [10]. In general, these systems use modulations, such as pulse position modulation (PPM) [11–14], transmitted reference (TR) [15], or on-off keying (OOK) [16] that do not require a coherent receiver. Analog techniques (e.g., correlation) are performed in order to significantly reduce the ADC sampling rate and power requirements. The large discrepancy between signal bandwidth and data rate implies that energy efficient solutions will avoid a Nyquist converter operating at the RF signal bandwidth.

## 1.1 ADC Architecture Overview

Figure 1-2 shows resolution and sampling frequency (or twice the signal bandwidth for oversampling converters) of recently published ADCs. This plot shows the trend of decreasing resolution at higher sampling rates described in Walden’s seminal ADC survey [17]. Of the prominent ADC architectures,  $\Sigma\Delta$  and successive approximation register (SAR) converters are typically used for low-speed, high-resolution applications. Pipelined converters dominate at medium speeds and resolutions, and folding and flash ADCs achieve the highest sampling rates but with low resolution.

The flash topology, along with its interpolating and folding variants, has been the conventional choice [18–22] for the high-speed, low-resolution ADCs considered in this thesis. While flash can provide the highest throughput, it requires an exponential growth in the number of comparisons with the resolution. The ensuing complexity motivates the use of

binary search that minimizes the difference between the analog input and digital output, calculating one bit at a time. The DAC is usually a binary-weighted capacitor array that also serves as the input sampling capacitor. Its linearity is insensitive to parasitic capacitance to ground. To make higher resolutions feasible, sub-DACs reduce the size of the main capacitor array [30], and calibration algorithms compensate for offset and linearity errors [31]. SAR is now the most common architecture for Nyquist conversion at low speeds and medium to high resolutions [31–41]; however, the fastest commercially available SAR converters sample at less than 5 MHz [40, 41], and the speed is limited due to the long latency of a conversion, which requires at least one clock cycle for each of the bit decisions. Using a non-binary SAR can be used for accurate conversion in the presence of incomplete comparator settling, decreasing the time for each bit decision; however, extra bit decisions are required. This technique has increased the sampling frequency to 20 MHz [35] and even 40 MHz without Nyquist performance [33].

Early on, the complexity advantage of the SAR architecture versus flash was identified, and the first monolithic time-interleaved converter was designed in 1980 using multiple SAR ADCs with less area than a 7-bit flash converter [42]. Time-interleaving uses multiple slower ADCs sampling at equally spaced instants, which combine to form a higher rate ADC [21, 33, 42–62]. The next use of a time-interleaved or parallel SAR ADC was for reduced energy consumption. In 1994, a simple analysis of comparator power showed the SAR should be superior to both flash and pipelined ADCs [62]. This advantage over flash follows the reduction of the number of comparisons from an exponential to a linear function of resolution. The time for comparator autozeroing justified the SAR over the pipelined architecture. A SAR converter only needs to autozero once per conversion ( $b$  comparisons), whereas a 1-bit/stage pipelined converter must autozero for every bit decision. This analysis holds for higher resolutions but fails to take into account some of the additional overhead of a SAR ADC. In particular, based solely on comparator complexity, a SAR ADC is more energy efficient than flash down to the 2-bit level, but when the digital logic energy is included, this is no longer the case.

Very recently, the time-interleaved SAR architecture has re-emerged as a low power alternative to flash and pipelined ADCs for the high-speed, low resolution converters necessary for UWB [46, 63]. For these specifications, a major limitation is digital power; a SAR converter includes digital feedback in the critical path, which must operate at high speed and can be power hungry. Fortunately, the digital power and speed improves with the scaling of CMOS technology, making SAR a viable alternative to flash even at very low resolutions in advanced technologies. Many of the scaling issues associated with the pipelined ADC are not present in the SAR ADC. The only active analog component, the comparator, still requires large gain and bandwidth, but because it has no closed loop stability requirement, this gain can be achieved through cascaded stages and positive feedback.

A widely used metric to compare ADCs across resolutions and speeds is the Figure of Merit [46]

$$FoM = \frac{P}{2^{ENOB} 2f_{in}}, \quad (1.1)$$

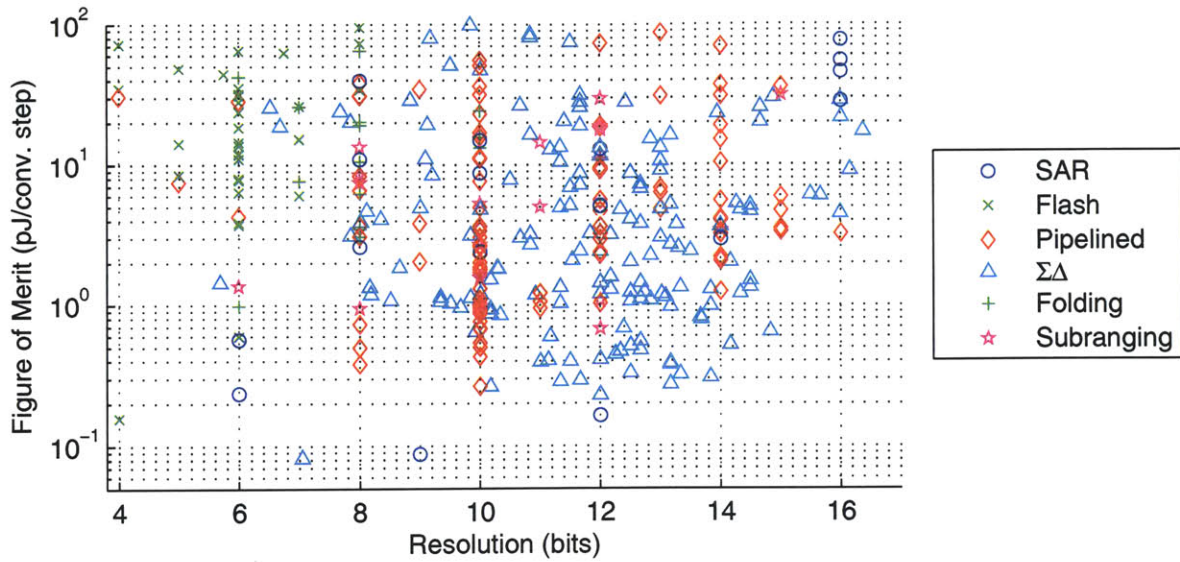
where  $P$  is the power dissipated, and  $ENOB$  is the effective number of bits of the converter, as defined in (1.2), at the input frequency  $f_{in}$ . Power tends to scale directly with the sampling rate and input bandwidth in an ADC, and exponentially with the resolution, making this  $FoM$  useful for comparing ADCs with different operating points. Superior energy efficiency corresponds to smaller  $FoM$ . In this thesis, the figure of merit is defined with  $f_{in}$  not to exceed the Nyquist frequency,  $f_s/2$ . The  $ENOB$  is the number of bits that would make an ideal ADC's signal-to-quantization-noise ratio (SQNR) equal to the measured ADC's signal-to-noise-plus-distortion ratio (SNDR), as in

$$SNDR = 6.02 \cdot ENOB + 1.76, \quad (1.2)$$

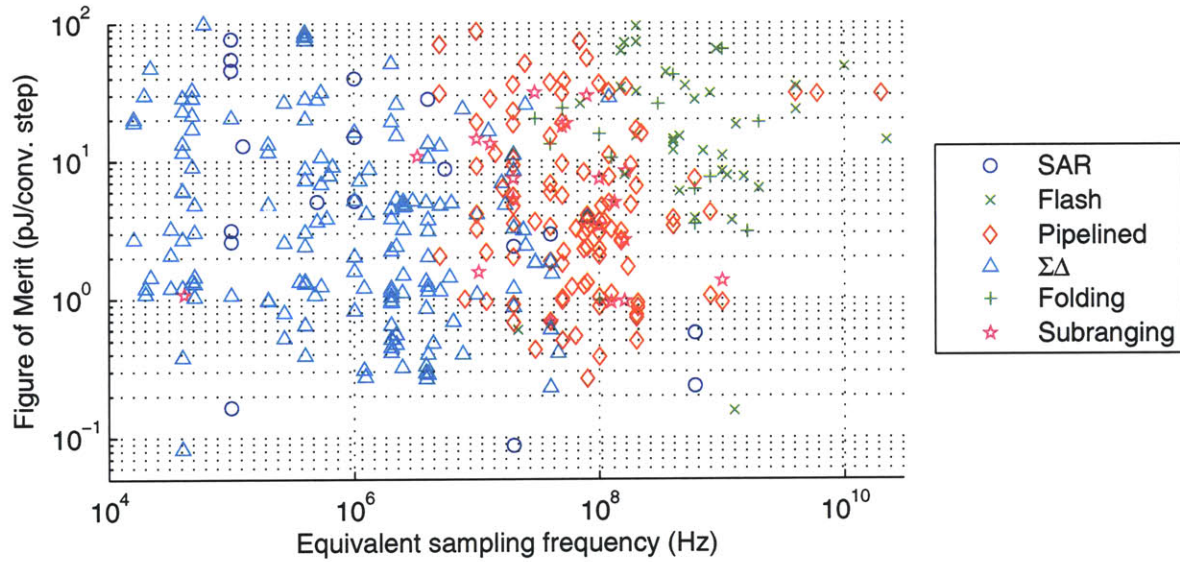
where SNDR is in decibels.

Other figures of merit, similar to (1.1), are also used. For instance, in [64], Scholtens and Vertregt use

$$FOM_{Scholtens} = \frac{P}{2^{ENOB_{DC}} 2ERBW}. \quad (1.3)$$



(a)



(b)

Figure 1-3: Plot of the ADC energy efficiency figure of merit (1.1) versus (a) resolution and (b) sampling frequency.

Here, the ENOB is determined at DC, and the effective resolution bandwidth (ERBW) is the frequency at which the ENOB drops by 0.5 bits from its low frequency value. By definition, (1.1), with  $f_{in} = ERBW$ , gives a  $\sqrt{2}$  larger figure of merit than (1.3). Other variants

replace  $2f_{in}$  with the sampling rate  $f_s$  [17], or  $ENOB$  with  $b$ , the nominal resolution [65], but these fail to reflect actual ADC performance. Finally, for high resolution circuit-noise-limited ADCs, replacing  $2^{ENOB}$  with  $2^{2ENOB}$  more accurately reflects the power scaling with resolution. This figure of merit has not gained widespread acceptance and would not apply for the low resolution ADCs that are the main study herein.

Figure 1-3 shows the values of (1.1) plotted against resolution and sampling frequency. The best converters have figures of merit lower than 100 fJ/conversion step and tend to be at lower resolutions. The rise in figure of merit on the right-side of Fig. 1-3(a) reflects the true power scaling of high-resolution ADCs as reflected in the  $2^{2ENOB}$   $FoM$ . In addition, the figure of merit is relatively independent of sampling frequency up until 1 GHz, although it should be noted that at the higher sampling frequencies, the resolution drops for the most energy-efficient converters. While the numbers of high frequency SAR converters is small (Fig. 1-2), Fig. 1-3(b) shows that they offer competitive energy efficiency at sampling rates in the hundreds of megahertz.

## 1.2 Thesis Contributions

This work focuses on the design of high-speed, low-resolution ADCs, with a particular emphasis on their implementation in deep sub-micron CMOS. As the target application is ultra-wideband radio, the particular specifications are 500 MS/s with 5 bits of resolution; thus, the ADC could be used in a receiver that uses the MIT pulse-based architecture, or any architecture that uses the minimum FCC bandwidth. The main contributions are investigation of the use of the SAR architecture for energy-efficient high-speed applications, joint design of the analog and digital circuits, techniques to facilitate highly parallel ADCs, and the mitigation of random fluctuations that are expected to get worse in advanced processes.

While flash is the typical architecture chosen to meet the target specifications, the energy-efficient performance of the SAR architecture in high-speed applications has previously been demonstrated [46]. Chapter 2 presents a comprehensive energy model to compare the power consumption of the time-interleaved SAR versus flash architectures. Chapters 3 and 4 present

two testchips that explore the benefits and limitations of high-speed SAR ADCs and the scaling of both the digital and analog circuitry into deep sub-micron CMOS. These chips feature a full custom digital controller and a new split capacitor array structure, respectively.

A commonly overlooked problem in high-speed mixed-signal circuits is the power consumption and latency of digital circuits, particularly those in series with the critical settling of analog signals. Joint timing design of the analog and digital sections can lead to increased settling time available for analog circuits and ultimately lower power consumption. Furthermore, an optimum mixed-signal voltage supply can be found where the overall energy per conversion is minimized. Chapter 5 defines the mixed-signal optimum energy point and describes the model used to fully optimize the ADC. For this joint analog/digital design, the SAR architecture is an excellent test case, as it uses comparable digital and analog power consumption, even in 65 nm CMOS technology, and digital logic is directly in the critical path of feedback between individual bit decisions.

The design of a near-optimally interleaved ADC is presented in chapter 6. For medium numbers of parallel channels (6–8), both the energy overhead of time-interleaving and performance degradation are minimal; however, to operate closer to the optimum supply voltage, significantly more channels are needed and many problems become evident. Clocking is one of the main sources of overhead because multiple phase-shifted low frequency clocks are required. In addition, the timing skew begins to limit high frequency performance and must be mitigated.

As feature sizes shrink, certain effects that are statistical in nature, such as threshold voltage mismatch from random dopant fluctuations, are increasing. To design a single 5-bit SAR converter channel in 65 nm technology is straightforward and well within the process limitations, but as increasing numbers of channels are integrated, they must be all designed for much more stringent specifications to ensure adequate overall yield. By characterizing these yield effects and using channel-level redundancy, significant yield improvements are achieved with minimal area and/or power cost.





## Chapter 2

# Parallelism in Voltage or Parallelism in Time

A quantitative energy model is developed to select between the SAR and flash architectures. This model is centered around the target 5-bit resolution, although it is valid over a small range of resolutions around it. This model significantly extends the simple comparator-centric model in [62] by including the remaining blocks in both architectures. One of the critical assumptions of this model is that any circuit noise is negligible in comparison to the quantization noise power, and the performance is therefore limited by static quantities, such as linearity and offset.

A generic SAR ADC, shown in Fig. 2-1(a), consists of a sample-and-hold circuit (S/H), a digital-to-analog converter (DAC), a comparator, and the successive approximation register control logic. The SAR algorithm determines each output bit sequentially from most significant bit (MSB) to least significant bit (LSB) in order to minimize the difference between the held analog input voltage and the output of the DAC. In the prevalent charge redistribution SAR ADC, the function of the S/H, DAC, and subtraction circuit is combined into a capacitor array, as shown in Fig. 2-1(b). For a charge redistribution SAR ADC, the comparator is the primary contributor to offset. The static nonlinearity is dominated by matching of the capacitor array and finite settling time of preamplifiers in the comparator. Signal-dependent

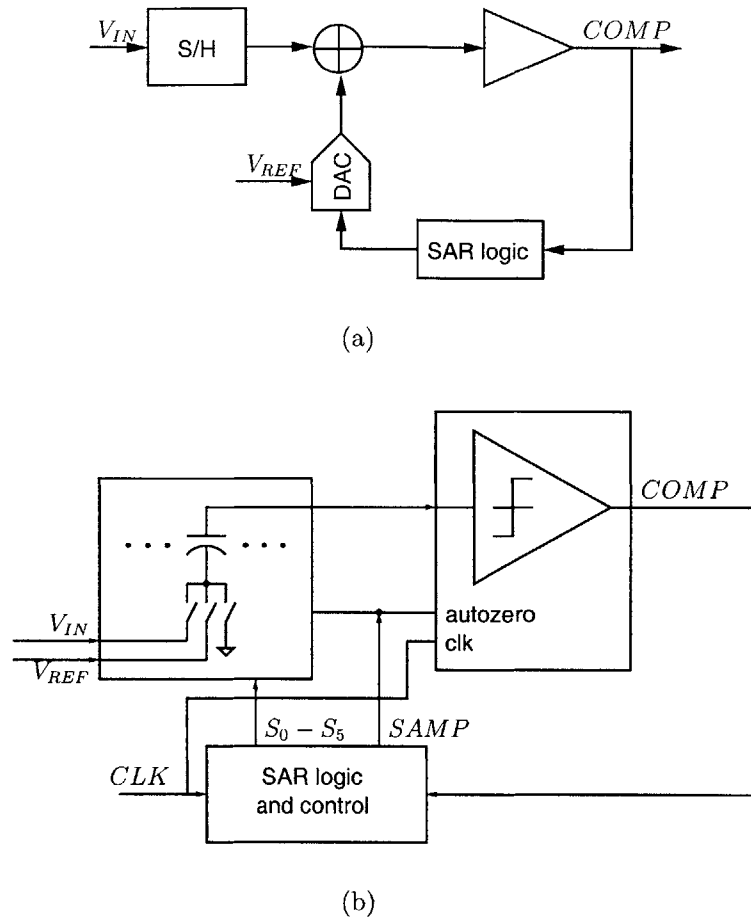


Figure 2-1: SAR ADC block diagram (a) with an explicit sample and hold circuit. (b) A single charge redistribution capacitive array combines the DAC, S/H and differencing circuit.

charge injection in the S/H can also contribute to static nonlinearity, but this is sufficiently cancelled at this low resolution with a differential implementation and the use of dummy switches. The dynamic linearity is dominated by the settling of the S/H network.

The  $b$ -bit flash ADC is shown in Fig. 2-2. A resistive ladder drives a bank of  $2^b - 1$  comparators, whose thermometer encoded output is converted to a binary code with a thermometer-to-binary encoder. Technically, a S/H is not required in a flash ADC, but the dynamic linearity is then limited by matching of the timing paths to all of the comparators, and, in practice, a S/H is frequently used. Static nonlinearity in a flash ADC is dominated by offsets in the comparators and matching in the resistive ladder. Dynamic nonlinearity

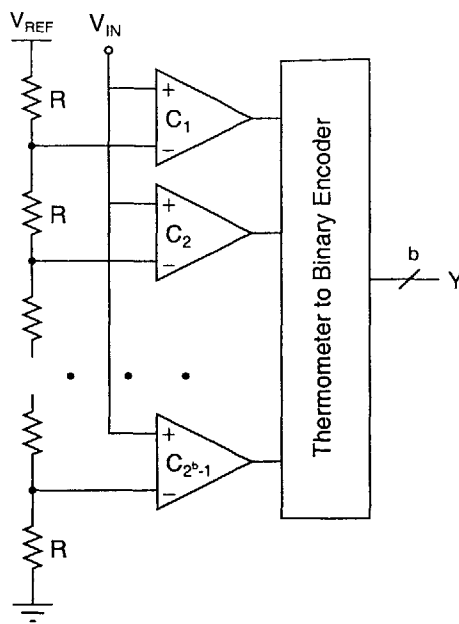


Figure 2-2: Flash ADC block diagram.

has a component from the S/H network and from finite settling time of the comparators.

In the discussion below, the contribution of the sampling network to dynamic (and static) nonlinearity is assumed identical in both the SAR and flash ADCs and is therefore neglected. This assumption is reasonable if the required sampling capacitance is the same in both architectures. The sampling capacitance in the SAR ADC is set by the total size of the capacitor array, and the sampling capacitance of the flash ADC is limited by the total input capacitance of the comparators. A separate sampling capacitor can be used to significantly reduce the input loading of the SAR ADC [38], but this is not suitable for a differential implementation. At the 5-bit level, the sampling capacitance of the SAR ADC is 320 fF, and the sum of the flash comparators' input capacitance is 400 fF. The total sampling network power is less than 10% of the overall power (cf. Fig. 5-10), and its difference between the two architectures is less than 2%, sufficiently small to ignore.

The SAR energy model assumes that converters are time-interleaved to equal the flash ADC's throughput. The number of parallel SAR converters,  $M$ , is constrained to be  $M = b + U$ , where  $b$  is the resolution and  $U$  is the number of clock periods used to sample the input and autozero the comparators. This constraint simplifies implementation and clock

distribution as discussed in Section 3.1. In particular, with this constraint, only one clock at the sampling frequency is required; thus, the clock generation requirements are the same for both a flash and a SAR ADC.

## 2.1 Component Energy Models

The energy requirements of each of the blocks used in a SAR and flash ADC is presented below. The constants are derived from simulations in a 0.18- $\mu\text{m}$  CMOS technology.

### 2.1.1 SAR Control Logic

The control logic for a SAR converter is based on a shift register of width  $b$  and consumes energy that grows approximately with  $b$ . For a given logic style that does not draw static current (e.g., static CMOS, dynamic, etc.), the total energy consumed by the switching of the control logic over one conversion is

$$E_{logic}^{SAR} \approx b C_{SWeq}^{SAR} V_{DD}^2 \quad (2.1)$$

where  $C_{SWeq}^{SAR}$  is the total switched capacitance normalized to the 1-bit level.

In a practical implementation, the total energy is expected to grow faster than  $b$ . The digital logic directly drives the switches in the capacitor array, the sizes of which must increase with the resolution to ensure sufficient settling time of the larger capacitor array. Thus, the total digital energy is the sum of the internal switching energy, which grows linearly with resolution, and the fanout energy to drive the switches. This fanout is relatively small and neglected here for the sake of keeping this energy comparison simple, but it is analyzed in more detail in Section 5.3.1.

## 2.1.2 Capacitor Array DAC

The capacitor array is a set of  $b$  binary-scaled capacitors and an extra unit capacitor. The well-known procedure to switch the array can be found in [66]. During the bit-cycling, an amount of charge proportional to the size of the array and the full-scale input voltage is switched onto the array. Assuming that this charge is supplied by a linear regulator or buffer connected to the analog voltage supply  $V_{DDA}$ , the total array energy per conversion is

$$E_{array} = 2\eta 2^b C_0 V_{DDA} V_{FS}. \quad (2.2)$$

The total energy drawn is input-signal dependent, which can be modeled using  $\eta(V_{IN})$  (cf., Fig. 4-9). Choosing  $\eta = 0.7$  gives the average current draw for a uniform input distribution across the full scale input range. The initial factor of 2 arises from the use of a fully differential capacitor array.

The unit capacitor size  $C_0$  is chosen to meet the linearity specification. The expected worst case linearity error occurs during the first bit-cycle, with a ratio error of

$$\frac{\Delta C}{C} = \frac{1}{\sqrt{2^{b-1}}} \frac{\Delta C_0}{C_0}, \quad (2.3)$$

where  $\Delta C_0$  representing the standard deviation of the unit capacitance. In order to maintain this error below the level of the least significant bit (LSB),  $\Delta C_0/C_0$  scales in proportion to  $1/2^{b/2}$ . Noting that  $\Delta C_0/C_0 \propto C_0^{-\zeta}$  with  $\zeta$  equal to 3/4 or 1/2 if the capacitance mismatch is dominated by edge effects or oxide variation, respectively, [67] the total array energy for one conversion is

$$E_{array}^{SAR} = 2\eta 2^{(1+1/2\zeta)b} \frac{C'_0}{2^{b'/2\zeta}} V_{DD} V_{FS} \quad (2.4)$$

where  $C'_0$  is the process-dependent capacitance required for matching to the  $b'$ -bit level.

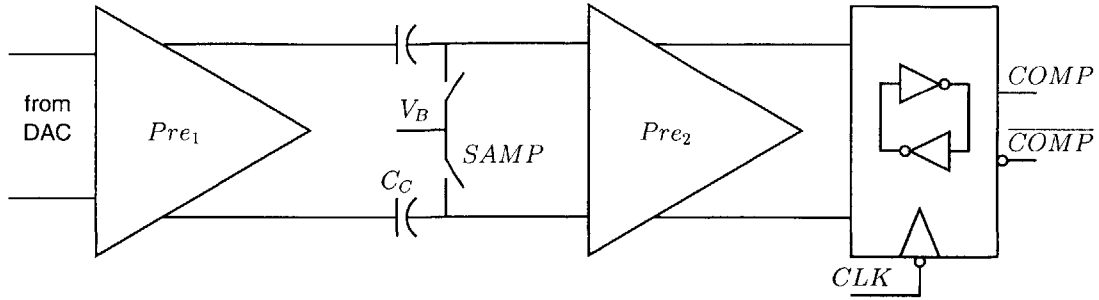


Figure 2-3: Comparator with a two-stage preamplifier and regenerative latch. Autozeroing of the first preamplifier occurs while the input is being sampled on the capacitor array ( $SAMP$  high).

### 2.1.3 Comparator

The comparator is characterized by its offset, noise, and speed, which includes both its total delay and the speed at which it can recover from a step change in input. Regenerative amplifiers have the best power delay product for a required amount of gain [68], but they are characterized by relatively large input referred offsets and are not well suited to simple offset cancellation methods. Typically, preamplifiers are inserted in series with the latch to amplify the comparator's input beyond the offset of the latch,  $V_{OL}$ . The preamplifiers' offset can be cancelled using output offset storage (OOS), as shown in the comparator schematic in Fig. 2-3. Here, two stages have been chosen in order to provide sufficient gain (9 – 25) for the expected  $V_{OL}$  and full scale input voltage,  $V_{FS}$ . Low gain per stage eases implementation in deep sub-micron CMOS. The preamplifiers also insulate the input of the comparator from kickback noise associated with the large swings at the output of the latch.

The preamplifier biases are chosen to satisfy four specifications: offset, noise, gain, and speed. The offset and flicker noise of the first preamplifier are eliminated by OOS, and it is assumed that the first preamplifier provides sufficient gain to compensate for the offset of the second preamplifier. For the high-speeds and low resolution required for this ADC, the currents required to meet the gain and settling time specifications are sufficient to meet the thermal noise requirements, and therefore only gain and speed are considered. Assuming that the minimum signal that must be reliably compared is one half of the LSB step size,

then the required gain is

$$A_V = A_{V1}A_{V2}\frac{C_C}{C_C + C_{in2}} = \nu A_{V1}A_{V2} = \frac{V_{OL}}{V_{OSin}} \approx \frac{2^{b+1}V_{OL}}{V_{FS}}. \quad (2.5)$$

Here,  $C_C$  is the OOS capacitor and  $C_{in2}$  is the input capacitance of the second preamplifier.

The preamplifiers are conservatively assumed to settle as a first order system, and must settle from the largest ( $V_{FS}$ ) to smallest input ( $V_{FS}/2^{b+1}$ ) in one half the clock period; the other half is used for latch resolution in a SAR ADC, or for autozeroing the preamplifiers between conversions in a flash ADC [62]. The settling time of the preamplifiers is

$$\tau = \tau_1 + \tau_2 = R_{L1}C_{L1} + R_{L2}C_{L2} = \frac{1}{N_\tau \cdot (2f_S)} \approx \frac{1}{(b+1)f_S \cdot 2 \ln 2}, \quad (2.6)$$

with  $N_\tau$  the number of time constants required for complete settling of the preamplifiers.

The approximate requirements for the the input referred offset and settling time of the preamplifiers, (2.5)–(2.6), are verified through behavioral simulations. Appendix A describes the setup for the behavioral simulations that are used throughout this thesis. Figure 2-4 compares the degradation in SNDR versus offset and settling time for the two architectures. The input referred offset requirement is slightly more stringent for a SAR ADC, but for a 1 dB drop in SNDR, the input referred offset standard deviation is approximately  $V_{LSB}/10$  (or  $3\sigma$  offset close to  $V_{LSB}/4$ ), smaller than the hand approximation. Figure 2-4(b) shows that, for a similar 1 dB drop in SNDR, the preamplifiers must settle for about 3 time constants in half a clock period, with close correlation between the two architectures; the theory behind (2.6) requires roughly 4 time constants.

Now, the individual current requirements can be calculated. Each of the preamplifiers has a differential NMOS input pair with resistive loads, as shown in Fig. 2-5, assumed to be loaded only by the input capacitance of the following stage. Using an approach similar to [69], both preamplifiers are biased at the same current density assumed fixed in this model<sup>1</sup>. Thus, there is a linear relationship between transconductance, input capacitance, and bias current

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<sup>1</sup>The actual current density is chosen as discussed in Section 3.2.3

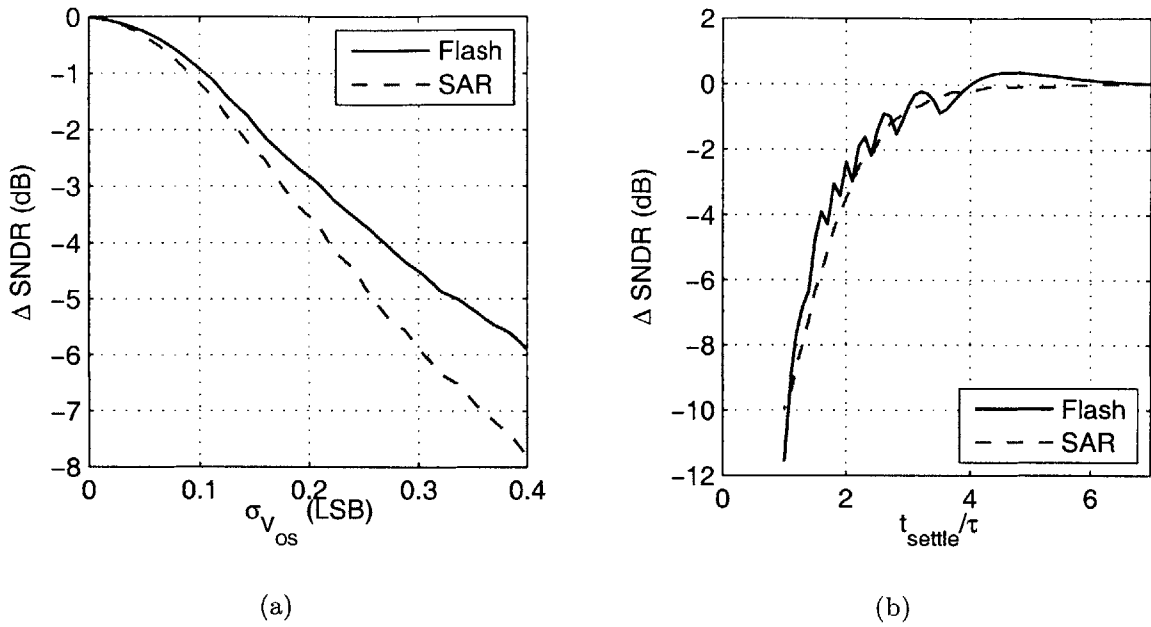


Figure 2-4: Behavioral simulation showing degradation in SNDR for the 5-bit time-interleaved SAR and flash architectures versus (a) input referred offset and (b) allowed settling time.

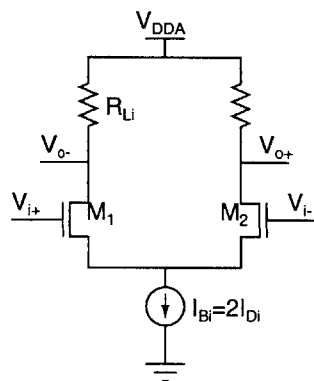


Figure 2-5: Basic preamplifier schematic.



( $\alpha g_m = I_D = I_B/2$ ,  $C_{in} = \beta I_D$ ), set by process and current density dependent parameters  $\alpha$  and  $\beta$ . Using  $A_{V_i} = g_{mi} R_{Li}$ , each preamplifier's bias current is

$$I_{Bi} = 2I_{Di} = 2\alpha g_{mi} = 2\alpha \frac{A_{Vi} C_{Li}}{\tau_i} \quad i = 1, 2 \quad (2.7)$$

The load capacitances for the two preamplifiers are  $C_{L1} = C_{in2} C_C / (C_{in2} + C_C) = \nu C_{in2}$  and  $C_{L2} = C_{ilatch}$ , the latch's input capacitance. The total preamplifier current is

$$I_{pre} = 2\alpha \frac{A_{V2} C_{ilatch}}{\tau_2} \left( 1 + \alpha \beta \nu \frac{A_{V1}}{\tau_1} \right). \quad (2.8)$$

Note that  $\alpha \beta = C_{ini} / g_{mi} \approx 1 / (2\pi f_T)$ , where  $f_T$  is the cutoff frequency of the transistor, and  $A_{V1} / \tau_1$  represents the unity gain bandwidth of the first preamplifier. Thus, the second addend within the parentheses in (2.8) represents how close the required unity gain bandwidth of the preamplifier is to the  $f_T$  of the device. The current rises rapidly as the operating frequencies approach the device  $f_T$ . For lower power consumption, it is desired to operate well below  $f_T$  ( $\alpha \beta \nu A_{V1} \ll \tau_1$ ), in which case (2.8) simplifies to

$$I_{pre} \approx 2\alpha \frac{A_{V2} C_{ilatch}}{\tau_2}. \quad (2.9)$$

To minimize the current,  $A_{V2}$  should be minimized and  $\tau_2$  should be maximized; however, that would cause the constraint  $\alpha \beta \nu A_{V1} \ll \tau_1$  to be violated. In addition using low gain stages was necessary to implement the preamplifiers in deep sub-micron processes with limited output resistance and low voltage supplies. Therefore, a near optimal result is to fix  $A_{V1} = A_{V2}$  and  $\tau_1 = \tau_2$ . Substituting (2.5) and (2.6) into (2.9) produces

$$\begin{aligned} I_{pre} &= 2\alpha \sqrt{\frac{V_{OL}}{\nu V_{OSin}}} \cdot 4N_\tau f_S C_{ilatch} \\ &\approx \frac{\alpha^2 \ln 2}{\nu^{1/2} V_{FS}^{1/2}} \cdot b 2^{b/2} f_S \sqrt{V_{OL}} C_{ilatch} \\ &= \xi b 2^{b/2} f_S \sqrt{V_{OL}} C_{ilatch}. \end{aligned} \quad (2.10)$$

The approximation in the second line assumes that  $V_{OSin} = V_{LSB}/4$  and  $N_\tau = b \ln 2$ , similar to the values derived by the behavioral simulations (Fig. 2-4).

The last two terms in (2.10) represent the preamplifier current dependence upon the latch. The offset and input capacitance of the latch is related by the matching properties of transistors. When threshold voltage variation is the dominant source of mismatch,  $V_{OL} \propto 1/\sqrt{A_{latch}} \propto 1/\sqrt{C_{ilatch}}$ , where  $A_{latch}$  is the area of the latch transistors [70]. Thus, the product term  $\sqrt{V_{OL}}C_{ilatch}$ , and correspondingly the preamplifier current, can be minimized by designing a latch with smaller transistors and larger input referred offsets. This tradeoff is limited by the maximum gain available from the two-stage preamplifier.

The total comparator energy per conversion, including the contribution of the latch, is

$$E_{comp}^{SAR} = \frac{b+U}{f_S} I_{pre} V_{DDA} + b C_{latch} V_{DD}^2 \quad (2.11)$$

for a SAR ADC, and it is

$$E_{comp}^{flash} = (2^b - 1) \left( \frac{I_{pre} V_{DDA}}{f_S} + C_{latch} V_{DD}^2 \right) \quad (2.12)$$

for flash. In both,  $V_{DDA}$  and  $V_{DD}$  are the analog and digital voltage supplies, respectively, and  $C_{latch}$  represents the total switched capacitance in the latch, which scales in proportion with its input capacitance. Conveniently, increasing the latch offset to reduce comparator bias currents also reduces the latch energy.

### 2.1.4 Resistor Ladder

Mismatch in the resistor ladder directly adds to any offset in the comparator to introduce static nonlinearity. Unlike capacitors, however, resistors do not have a direct relationship between resistor value and matching. For a given sheet resistance  $R_\square$  the resistor's value is determined only on its aspect ratio ( $R = (W_R/L_R)R_\square$ ), but the matching is related to only total area and perimeter. Thus, matching of a fixed resistor can be improved simply by scaling up both its width and length, keeping the aspect ratio constant. Dynamic linearity

requirements dictate the actual resistance value because the outputs of the resistance ladder must be sufficiently stable against kickback and input feedthrough. The kickback from the latch is made negligible with the two stage cascaded preamplifiers, and the maximum value of the resistor ladder is then set by input feedthrough, whose maximum value at the middle point of the resistor ladder is [71]

$$\frac{V_{err}}{V_{FS}} = \frac{\pi}{4} f_{in} R_{ladder} C_{feedthrough}. \quad (2.13)$$

Here,  $C_{feedthrough}$  is the total feedthrough from the input through all of the comparators to the reference ladder and is roughly  $2^b C_{in1}/2$ . The total resistance in the resistor ladder is  $R$ , and  $f_{in}$  is the maximum input frequency. Assuming that a maximum error of  $1/2V_{LSB}$  is acceptable at a Nyquist input frequency, the total reference ladder power is

$$P_{res-ladder} = \frac{V_{REF}^2}{R_{ladder}} = 2^{b+1} \frac{\pi}{4} 2^{b-1} C_{in1} V_{FS}^2. \quad (2.14)$$

### 2.1.5 Thermometer-to-Binary Encoder

The outputs of the comparators ideally form a thermometer encoded number, where, for a digital output code  $d$ , the bottom  $d$  comparators' outputs are all 1, and the upper  $2^b - d$  comparators all output 0. When offsets and dynamic effects are taken into account, the output of the comparators can exhibit bubble codes, where a 0 is inserted within the sequence of 1s, giving an ambiguous digital output. Thermometer-to-binary encoder design thus entails minimizing the digital output error in the presence of typical bubble errors. At the limit, the encoder can be built using an adder tree, adding up all of the 1's, independent of location, to determine the digital output. Any single bubble error then produces a maximum 1-bit error on the output. The principal disadvantage of this "ideal" encoder is that its complexity scales faster than exponentially with resolution. A Wallace tree adder implementation scales as roughly  $O(b2^b)$  [72]. Other implementations trade off robustness to arbitrary bubble errors with complexity [73–76]. At a minimum, the size and power dissipation of the thermometer-to-binary encoder must scale with the number of comparators, leading to an

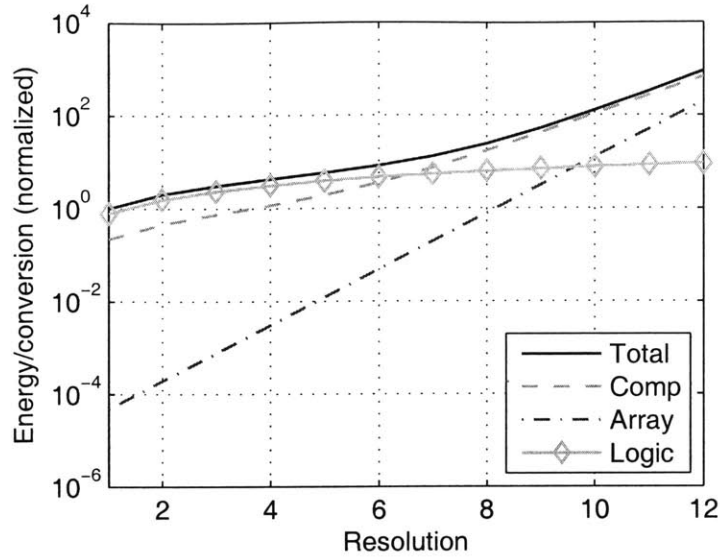


Figure 2-6: Theoretical SAR energy versus resolution, along with the individual comparator, array, and logic components, all normalized to the 1-bit level.

energy of

$$E_{T \rightarrow B} \approx 2^b C_{T \rightarrow B,0} V_{DD}^2, \quad (2.15)$$

where  $C_{T \rightarrow B,0}$  is the basic thermometer-to-binary encoder element's switched capacitance.

To be fair to the flash architecture, the comparison included herein only equates the throughput of the flash and SAR topologies, but the flash latency is still shorter. Therefore, pipelining and reduced supply voltage can be used to further reduce the power of the thermometer-to-binary encoder. At 5-bits, the thermometer decoder energy at the full  $V_{DD}$  is less than 10% of the overall flash power (cf. Fig. 2-7), much less than the energy difference between the SAR and the flash topologies, so any energy improvement obtained by pipelining the thermometer-to-binary encoder would not change the final architectural selection of the model.

## 2.2 Composite Energy

Summing (2.11), (2.4), and (2.1), the total SAR energy per conversion is

$$\begin{aligned}
 E_{SAR} = & \xi b(b + U)2^{b/2} \sqrt{V_{OL}} C_{ilatch} V_{DDA} \\
 & + 2\eta 2^{(1+1/2\zeta)b} \frac{C'_0}{2^{b'/2\zeta}} V_{DD} V_{FS} \\
 & + b(C_{latch} + C_{SWeq}^{SAR}) V_{DD}^2.
 \end{aligned} \tag{2.16}$$

A plot of (2.16) versus resolution is shown in Fig. 2-6, where three different regions are clearly seen. At low resolutions, the digital energy dominates, and the energy grows linearly with  $b$ . At some point, the comparator begins to dominate, with energy growing as  $b^2 2^{b/2}$ . Finally, at high resolutions, the growing size and matching requirements of the capacitor array dominate, and the energy grows as  $2^{(1+1/2\zeta)b}$ ; however, the model is valid for at most 7 bits of resolution due to the gain limitation in the preamplifier and the neglect of noise and other non-idealities.

The total flash energy per conversion, calculated using (2.12), (2.14), and (2.15), is

$$\begin{aligned}
 E_{flash} = & 2^b \left( \xi b 2^{b/2} \sqrt{V_{OL}} C_{ilatch} V_{DDA} \right. \\
 & + (C_{latch} + C_{T \rightarrow B,0}) V_{DD}^2 \\
 & \left. + \frac{\pi}{4} 2^b C_{in1} V_{FS}^2 \right),
 \end{aligned} \tag{2.17}$$

and it is plotted in Fig. 2-7.

## 2.3 Flash Variants

Two variants of the flash architecture, interpolating [22, 77–80] and folding [59, 71, 81, 82] reduce the number of preamplifiers and latches, respectively. The former uses resistors connected between the outputs of neighboring preamplifiers to interpolate outputs that are between the two preamplifier trip points (Fig. 2-8), reducing the number of preamplifiers

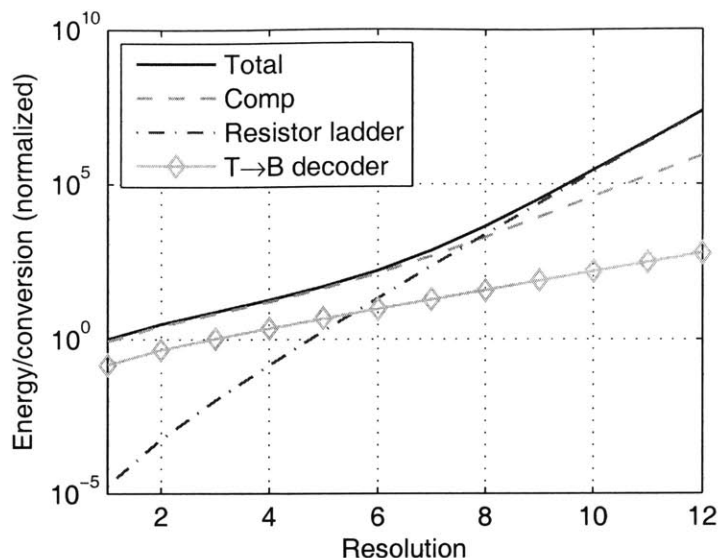


Figure 2-7: Theoretical normalized flash energy versus resolution, including the individual contributions of the comparator, resistor ladder, and thermometer-to-binary encoder.

needed to drive the latches. In addition, the same resistors can average the offsets of neighboring preamplifiers, improving differential nonlinearity [78, 83]. As shown, if the resistors are to be effective for averaging, they also act to lower the gain of the preamplifiers, thereby increasing the input referred offset of the latches; however, this limitation has been circumvented by replacing the resistors connected to the supply with high impedance current source loads [78]. In addition, the capacitive load of the preamplifiers is increased because of the load of multiple latches. Because the preamplifier current is linear with the latch's capacitive load (see (2.10)), the increase in load capacitance directly counterbalances the savings in the number of preamplifiers, and the overall energy per conversion would be identical. In practice, when effects such as parasitics and noise are considered, resistive interpolation can lower overall flash energy.

The folding technique reduces the number of latches required by generating a signal with multiple zero crossings at the output of the preamplifiers. Figure 2-9 shows a generalized folding amplifier, and the output of a fold by 4 block. The output currents of differential pairs are connected alternatively to the positive and negative outputs. Combined with a coarse quantizer to determine which of the regions (I–IV) the input is in, a single latch can

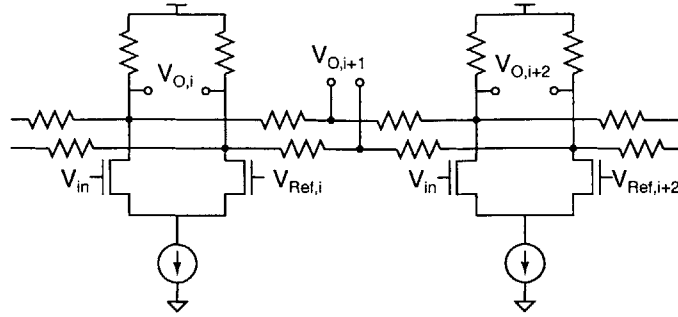


Figure 2-8: Diagram showing interpolation by 2 between outputs of neighboring preamplifiers. Additional taps can be placed between the outputs of successive preamplifiers to further reduce their total number.

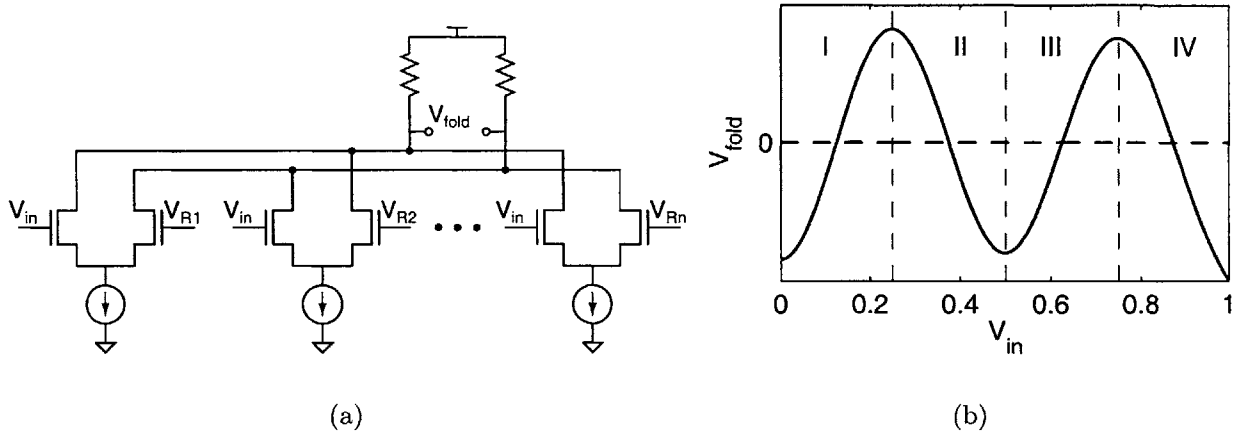


Figure 2-9: (a) Schematic of an amplifier with folding factor of  $n$ . (b) The simulated output of a fold-by-4 block.

now resolve four trip points. The tradeoff with folding is that the speed of the preamplifier block is now increased by the folding factor. Assuming a folding factor of  $m_{fold}$ , the total energy per conversion for a folding flash ADC is

$$\begin{aligned}
 E_{folding} = 2^b & \left[ \xi m_{fold} b 2^{b/2} \sqrt{V_{OL}} \frac{C_{ilatch}}{m_{fold}} V_{DDA} \right. \\
 & + \left( \frac{C_{latch} + C_{T \rightarrow B,0}}{m_{fold}} \right) V_{DD}^2 \\
 & \left. + \frac{\pi}{4} 2^b C_{in1} V_{FS}^2 \right]. \tag{2.18}
 \end{aligned}$$

Once again, due to the simple preamplifier current equation's linear dependence on the latch input capacitance, the increased speed requirement cancels the decreased load capacitance, leading to no change in preamplifier energy, but latch energy is reduced.

## 2.4 Architecture Comparison

A comparison of the theoretical SAR flash, and folding energies is plotted versus resolution in Fig. 2-10. Above 4 bits, the SAR ADC has the better energy efficiency than a flash ADC, whereas its digital overhead limits its performance at the lowest resolutions; however, the savings at 5-bit are slim (approximately 30%) and could disappear entirely without sufficient control of the digital power consumption. In addition, the folding flash ADC offers the best energy efficiency at very low resolutions, but at higher resolutions it is less energy efficient than both SAR and flash. The latter may be somewhat surprising because (2.18) is strictly less than (2.17). These equations, however, used the approximation  $\alpha\beta\nu A_{V1} \ll \tau_1$ , but for increased resolutions,  $\tau$  decreases,  $A_V$  increases, and the folding ADC with a required speed twice that of the conventional flash ADC, operates too close to the  $f_T$  of the devices, and the preamplifier current starts to increase quickly. By time-interleaving folding ADCs, the preamplifier requirements could remain well below the device limitations, and the slope of the folding energy would follow that of the flash and remain lower in energy up to higher resolutions, but, for the sake of clarity, this was not shown in the figure.

Finally, this energy model can be used to provide some insight into expected energy advantages of different design choices and process generations. The principal advantage that SAR has over flash is that it requires only  $b$  instead of  $2^b$  comparators. As the energy of an individual comparison decreases relative to the SAR digital power, flash will be more attractive. Conversely, as the digital power decreases, SAR will outperform flash. In deep sub-micron CMOS, the digital energy experiences the greatest benefit (both  $V_{DD}$  and  $C_{SWeq}^{SAR}$  shrink). The comparator energy also improves, as  $V_{DDA}$  and  $C_{latch}$  decrease but not at the same rate as the digital. One potential limitation is the reduction of  $V_{FS}$ , which increases preamplifier current (2.10).



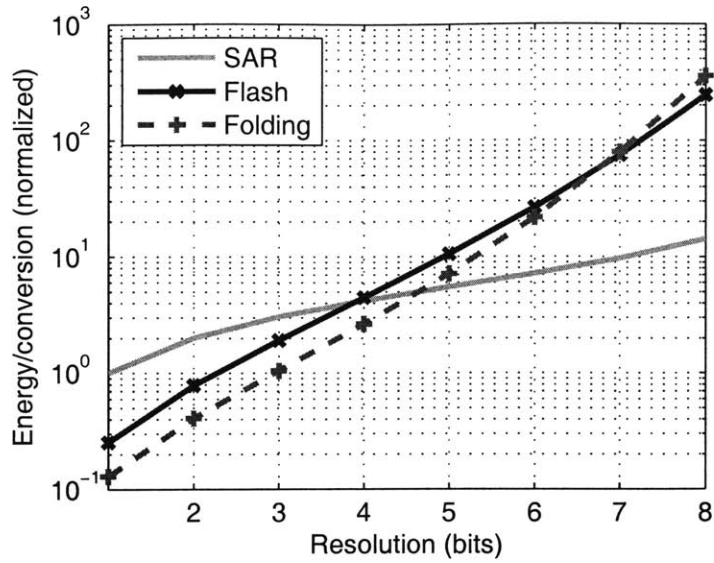


Figure 2-10: Modeled SAR and flash ADC energies versus resolution. The folding ADC uses a folding factor of 2.

At the 5-bit level, the time-interleaved SAR ADC is expected to outperform both flash and folding topologies. The remainder of this thesis presents the design and measurements of three test chips that demonstrate this architecture. A more comprehensive energy model is developed in Chapter 5, but the analysis presented above is the basic foundation behind the design of the first two test chips, with ideas presented as to how to lower the energy consumption of all three components: capacitor array, comparator, and logic.



# Chapter 3

## Initial Foray into Time-Interleaved SAR Design: Digital Challenges

The first prototype serves to explore the challenges in high speed implementations of the time-interleaved SAR architecture. As this prototype is the one actually used in the UWB high-data-rate system, two converters suitable for sampling of I/Q signals are incorporated, and a resolution scaling mechanism is included, whereby signal processing accuracy can be reduced to achieve overall system power savings under favorable wireless channel conditions. The principle challenge in the chosen technology, 0.18- $\mu\text{m}$  CMOS, is the integration of the digital logic controller, and a custom logic block with dynamic registers is developed that consumes only 20% of the power of a corresponding static CMOS implementation. This chip also features self-timed bit-cycling, an example of joint timing design between the analog and digital blocks. These features, as well a duty cycled comparator, are discussed in depth below, followed by measurements of the ADC.

### 3.1 Top-Level Implementation

Time-interleaved SAR has been chosen here because it is expected to be lower power than flash at 5 bits. The prototype contains two identical ADCs, a start signal generator, a

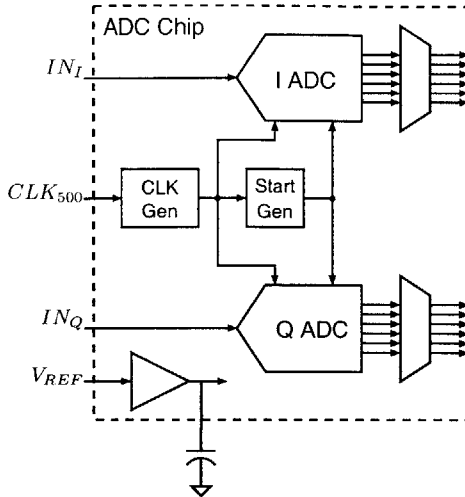


Figure 3-1: Simplified chip block diagram.

reference voltage buffer, and a clock generation block, as shown in Fig. 3-1. The reference voltage buffer is used to drive the bottom plate of the capacitor arrays with a reference voltage that is externally set between 100–500 mV. It must have sufficiently low output impedance to supply the roughly  $500 \mu\text{A}$  average current drawn by the capacitor arrays. An external  $0.1 \mu\text{F}$  decoupling capacitor is used to filter out the high frequency current pulses that are drawn at the sampling frequency. The clock generation unit contains a high-speed amplifier chain for use with a low-swing external clock.

To make the channel internal clock synchronous with the sampling clock, thereby minimizing clock distribution overhead, six time-interleaved channels are used for the 5-b converter (one of the assumptions in Chapter 2 with  $U = 1$ ). Therefore, only a single global clock is shared by all the channels, eliminating complex circuitry to produce multiple phase shifted clocks. A second benefit of this choice is that all critical sampling edges are aligned to the same shared clock; timing skew is minimized by careful layout matching of the input and clock paths. Each of the channels begins a conversion upon receiving a start signal generated by the previous channel. A global start signal triggers the first channel in both the I and Q ADCs on the chip. The top-level block diagram of one ADC is shown in Fig. 3-2; all channels share the same input, reference, and clock signals. A full clock period is devoted to the generation and propagation of the start path between channels, so time-interleaving

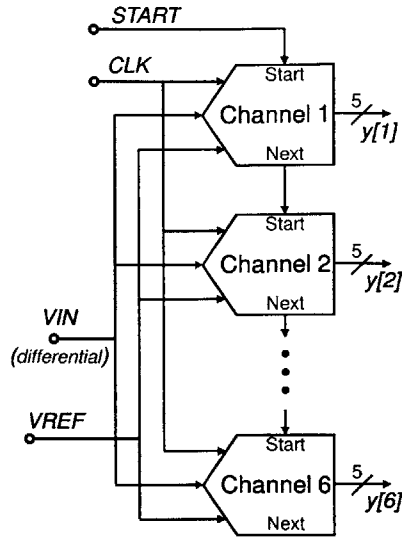


Figure 3-2: Top-level block diagram of a 6-way time-interleaved SAR ADC

and synchronization are achieved with minimal overhead.

Each channel, as shown in Fig. 3-3, is similar to the generic SAR ADC of Fig. 2-1(b), with the addition of signals for synchronization between channels and a *DONE* feedback signal from the comparator needed for self-timed bit-cycling, described below. The figure shows a single-ended path, but the actual implementation is fully differential. The circuit details of the channel sub-blocks are described in the next section.

## 3.2 Channel Circuits

### 3.2.1 Self-Timing

One disadvantage of the SAR architecture is the feedback required between successive clock periods. Specifically, the result of the previous comparison is necessary to generate an improved estimate for determining the next bit. While this feedback path is entirely digital, its latency must be minimized to allow maximum time for analog signals to settle in the DAC capacitor array and preamplifiers.

To decrease the latency during bit-cycling, self-timing is used [37], wherein the latch triggers the start of the next bit-cycle when it has resolved a value. In effect, after the latch

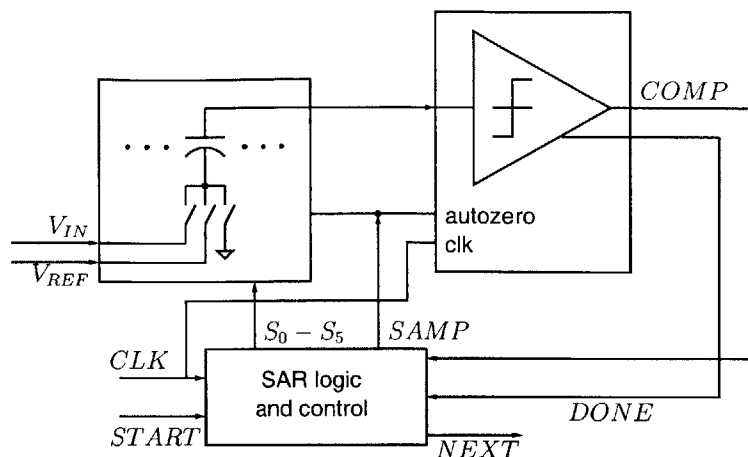


Figure 3-3: Channel block diagram with self-timing feedback loop and time-interleaving synchronization signals.

output has settled, the remainder of the second half of the clock period is borrowed by the DAC and preamplifiers for the next bit. Self-timing is particularly useful because the latch typically resolves in much less than 1 ns (half the clock period).

The timing of the ADC is shown in Fig. 3-4, which considers the sampling operation and the first three bit decisions. The assertion of  $SAMP$  is delayed from the rising edge of the shared system clock ( $CLK$ ) to avoid a high frequency current impulse on the analog inputs at the same time that the previous channel finishes sampling. During bit-cycling, the regeneration of the latch is always triggered by the falling edge of  $CLK$ . In the first bit decision, the latch resolves quickly, asserting the  $DONE$  signal (XOR of the latch outputs). This causes  $CLK_{BC}$ , the bit-cycling clock, to transition high immediately, rather than waiting for the rising edge of  $CLK$ . During the second period, the latch output is metastable and does not resolve. In this case, the rising edge of  $CLK_{BC}$  is triggered by  $CLK$ , and bit-cycling continues as normal. In simulation, self-timed bit-cycling extends the period given to settling of the capacitor array and preamplifiers by up to 20%, which is used to reduce the preamplifier currents and analog power by a similar amount.

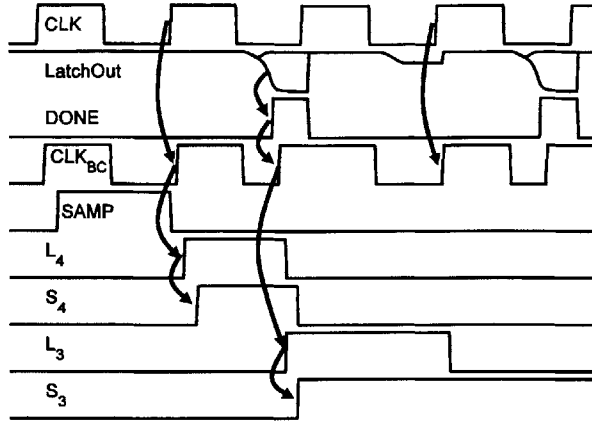


Figure 3-4: Timing diagram showing sampling and the first three periods of bit-cycling. In the first bit-cycle, the *DONE* signal triggers *CLK<sub>BC</sub>* to start the next bit-cycle, whereas the second period shows that bit-cycling is triggered by *CLK* when the latch is metastable.

### 3.2.2 SAR Logic

To implement a fast SAR controller, the logic, shown in Fig. 3-5, consists of a shift register ( $SR_0 - SR_5$ ) and a set of switch drive registers ( $SW_0 - SW_4$ ). This topology allows the larger switch drive registers to be clocked only once per conversion as opposed to every clock period like the smaller shift registers, while keeping the critical path delay to only  $2t_{c-q}$ , where  $t_{c-q}$  is the delay from the rising clock edge to valid outputs of a register. The current bit being resolved is determined by  $L_4 - L_0$ , only one of which is high every period.  $S_i$  asserts on the rising edge of  $L_i$  at the start of the decision period for bit  $i$ . When  $L_i$  falls, the decision for that bit is made, bringing  $S_i$  back low only if *COMP* is high.

Both the shift registers and the switch drive registers are implemented using full custom dynamic logic. The shift registers  $SR_0 - SR_1$  are conventional C<sup>2</sup>MOS registers, while  $SR_2 - SR_5$  have been modified as shown in Fig. 3-6. Transistors  $M_1 - M_4$  and  $M_9 - M_{12}$  form two C<sup>2</sup>MOS latches, with  $M_{13} - M_{14}$  producing the complementary output.  $M_5 - M_6$  provide a reset operation, ending bit-cycling whenever *START* is received by the channel. When this channel is not enabled (e.g., due to bit scaling, see Section 3.2.4),  $M_7 - M_8$  force  $L_i$  low.

The switch drive logic function is implemented in the single register shown in Fig. 3-7. Transistor  $M_1$  provides a fast charge path for  $S_i$  upon the rising edge of  $L_i$ .  $M_4 - M_7$  form

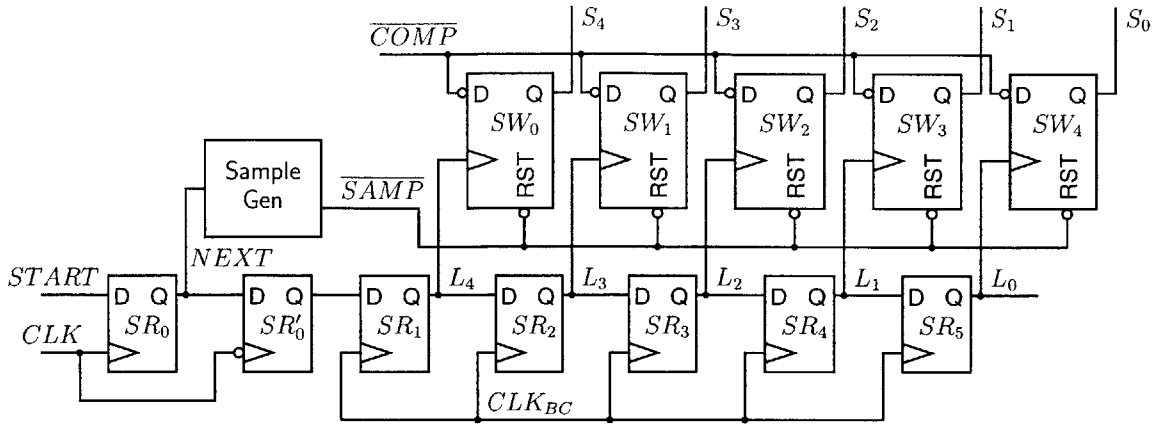


Figure 3-5: SAR logic implementation. The shift registers  $SR_0 - SR_5$  clock one of the switch drive registers  $SW_0 - SW_4$  during every phase of bit-cycling. The outputs  $S_0 - S_4$  control the DAC capacitor array.

a C<sup>2</sup>MOS latch that holds the value of  $\overline{COMP}$  after the falling edge of  $L_i$  and discharges  $S_i$  through  $M_2 - M_3$ . During sampling,  $M_{12} - M_{14}$  reset  $S_i$  low. Transistors  $M_8 - M_{10}$  are used to precharge node  $a$  to avoid charge sharing effects when  $M_2$  turns on. Finally,  $M_{11}$  puts the register in a known state when disabled.

Using this custom logic allows a faster and lower power SAR implementation. In simulations, the custom logic was 58% faster while consuming only 20% of the power of an equivalent SAR built using static CMOS and transmission gate feedback flip-flops.

### 3.2.3 Comparator Design

The comparator of Fig. 2-3 has been modified as seen in Fig. 3-8(a) by adding preamplifier enable signals and switches  $S_{p2}$ . The preamplifiers, shown in Fig. 3-8(b), have been designed using the equations presented in Sec. 2.1.3, with  $M_5$  added to enable duty cycling. PMOS transistors  $M_3 - M_4$  operate in the linear region and function as the loads  $R_L$ . The bias current density has been set by considering (2.8). Ignoring the second term, minimum energy corresponds to operating at low  $\alpha$  (high  $g_m/I_D$ ), which suggests operating in moderate to weak inversion. Unfortunately, the  $f_T$  of the transistors drops in moderate inversion, and a suitable current density in strong inversion has been chosen to enable the high operating



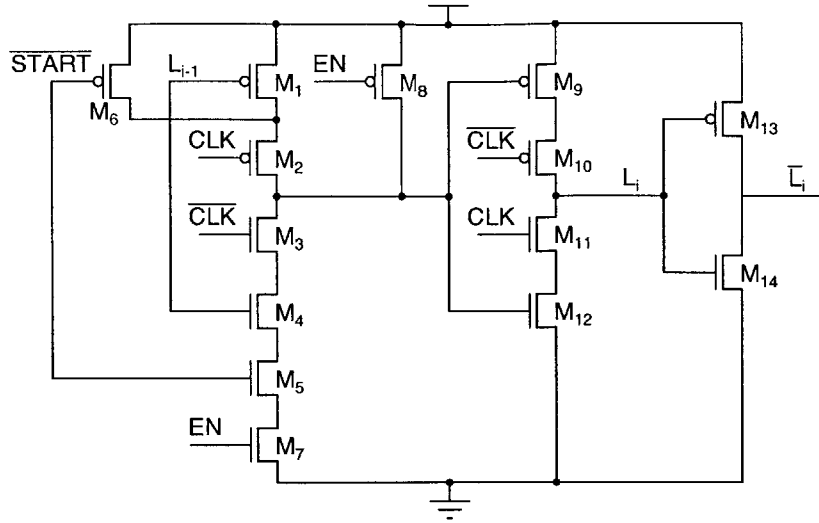


Figure 3-6: Resettable shift register built on C<sup>2</sup>MOS latch.

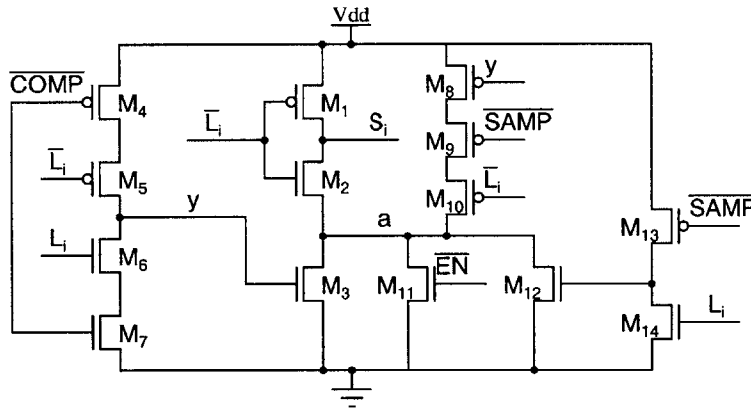
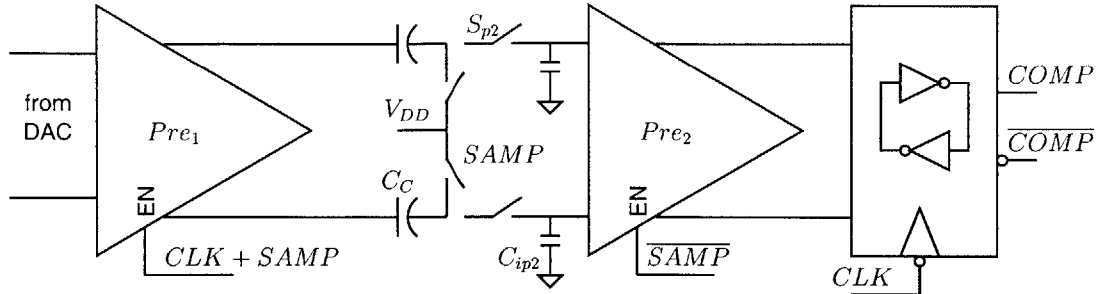


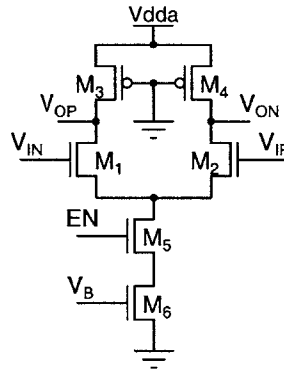
Figure 3-7: Schematic of the switch drive registers.

speeds ( $A_{V2}/(2\pi\tau) \approx 2\text{GHz}$ ). The preamplifiers' simulated frequency performance is shown in Fig. 3-9. The latch is a conventional sense-amplifier flip-flop [84].

In order to reduce average current draw, the preamplifiers are duty-cycled using  $M_5$  and  $S_{p2}$ , turning off their currents when appropriate. While sampling of the input, the output offset voltage of  $Pre_1$  is stored on capacitors  $C_C$ , while  $Pre_2$  is disabled. During the first half of every bit decision period,  $CLK$  is high (see Fig. 3-4) and the preamplifiers are actively settling. When  $CLK$  goes low, the preamplifiers are assumed to have settled and are holding their outputs steady; switches  $S_{p2}$  are opened, storing the first preamplifier's



(a)



(b)

Figure 3-8: (a) Comparator diagram with (b) duty-cycling preamplifier schematic.

output on  $C_{ip2}$ , the parasitic capacitance at the input to the second preamplifier, and the current through  $Pre_1$  is disabled. The energy savings are proportional to the amount of time that  $Pre_1$  is disabled; however, due to the self-timed bit-cycling, this might be very short. In this prototype,  $Pre_1$  is clocked by the 50% duty cycle  $CLK$ . The extra time from early switching of  $CLK_{BC}$  is used only for the settling of the capacitor array. Even though this extra time is not used directly by the preamplifiers, duty cycling does improve the settling time because after  $Pre_1$  is disabled by turning off  $M_5$ , both  $V_{OP}$  and  $V_{ON}$  float to  $V_{DDA}$ , yielding a zero differential output and effectively recovering from the previous output state.

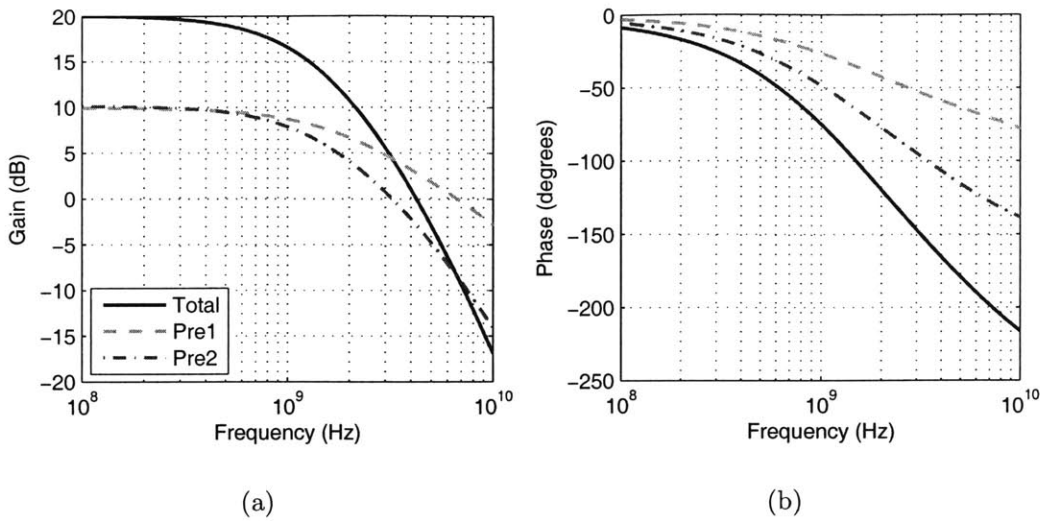


Figure 3-9: Simulated (a) gain and (b) phase of the preamplifiers.

### 3.2.4 Bit-scaling and I/O Circuits

For flexibility in response to varying system requirements, the ADC is capable of scaling its resolution  $b$  from 1 to 5 bits at a constant sampling rate. Since each channel takes  $b + 1$  periods to perform a conversion, fewer channels are needed to maintain the same throughput; the unused  $5 - b$  channels are disabled by stopping the propagation of the *START* signal, gating the clocks, and shutting off all bias currents. The start generation block triggers the first channel at a higher rate. When a channel receives the *START* signal, it samples during the next clock period, and the sampling operation automatically ends bit decisions at the correct, lower resolution.

To reduce circuit board requirements between the ADC and the digital back-end of the UWB transceiver, the outputs of the channels are not serialized, yielding an output frequency of 83 MHz. When not operating in the full 5-bit mode, some of the channels are deactivated. Rather than placing the burden on the back-end chip to sort the ADC outputs, a mux tree (see Fig. 3-1) is implemented to maintain a constant 83 MHz output rate independent of the resolution, with the unused LSBs of each word forced to 0. Level conversion circuits are placed between the core ADC and the I/O circuits to allow an internal digital supply lower

Time Per.	Ch 1	Ch 2	Ch 3	Ch 4	Ch 5	Ch 6
1	<i>S</i>	<i>BD0</i>	<i>BD1</i>	<i>BD2</i>	<i>BD3</i>	<i>BD4</i>
2	<i>BD4</i>	<i>S</i>	<i>BD0</i>	<i>BD1</i>	<i>BD2</i>	<i>BD3</i>
3	<i>BD3</i>	<i>BD4</i>	<i>S</i>	<i>BD0</i>	<i>BD1</i>	<i>BD2</i>
4	<i>BD2</i>	<i>BD3</i>	<i>BD4</i>	<i>S</i>	<i>BD0</i>	<i>BD1</i>
5	<i>BD1</i>	<i>BD2</i>	<i>BD3</i>	<i>BD4</i>	<i>S</i>	<i>BD0</i>
6	<i>BD0</i>	<i>BD1</i>	<i>BD2</i>	<i>BD3</i>	<i>BD4</i>	<i>S</i>

(a)

Time Per.	Ch 1	Ch 2	Ch 3	Ch 4	Ch 5	Ch 6
1	<i>S</i>	<i>BD2</i>	<i>BD3</i>	<i>BD4</i>	<i>OFF</i>	<i>OFF</i>
2	<i>BD4</i>	<i>S</i>	<i>BD2</i>	<i>BD3</i>	<i>OFF</i>	<i>OFF</i>
3	<i>BD3</i>	<i>BD4</i>	<i>S</i>	<i>BD2</i>	<i>OFF</i>	<i>OFF</i>
4	<i>BD2</i>	<i>BD3</i>	<i>BD4</i>	<i>S</i>	<i>OFF</i>	<i>OFF</i>
5	<i>S</i>	<i>BD2</i>	<i>BD3</i>	<i>BD4</i>	<i>OFF</i>	<i>OFF</i>
6	<i>BD4</i>	<i>S</i>	<i>BD2</i>	<i>BD3</i>	<i>OFF</i>	<i>OFF</i>

(b)

Figure 3-10: Table of states for all channels operating under (a) 5-bit and (b) 3-bit resolution. In the tables, *S* represents sampling, and *BD* is the bit decision phase for bit *i*. For the lower resolution modes, *OFF* means that the clocks are gated and all bias currents are shut down.

than 1.8 V while still maintaining the proper voltage for inter-chip communication.

### 3.3 Measured Results

The dual ADC chip was fabricated in a 0.18- $\mu\text{m}$  CMOS process with poly-poly capacitors. The photograph of the 5.5 mm<sup>2</sup> die is shown in Fig. 3-11. The active area of each ADC is 1 mm x 0.5 mm. With a 500 MHz sampling clock, the ADC exhibits an INL and DNL of  $-0.26/0.42$  and  $-0.39/0.33$  LSBs, respectively, as shown in Fig. 3-12 and measured using a code density test with a full swing sinusoidal input [85].

For dynamic testing, the Tektronix AWG710 arbitrary waveform generator was used to generate the sinusoidal input using an 8-bit 4 GS/s DAC. The AWG710 also provided the

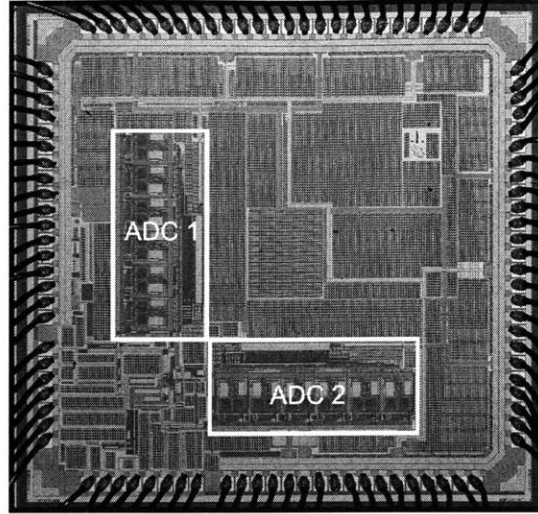


Figure 3-11: Die photograph of dual-ADC chip in 0.18- $\mu\text{m}$  CMOS. Total active area is 1.1 mm<sup>2</sup>.

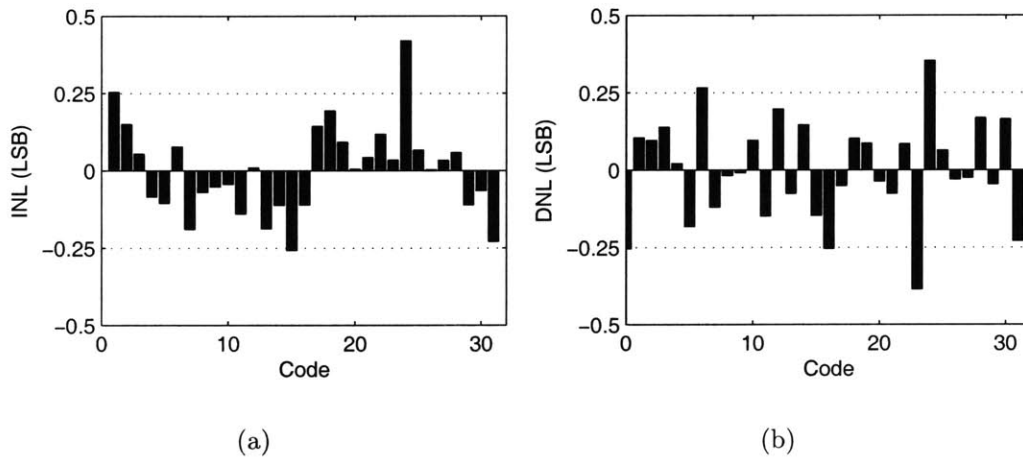


Figure 3-12: Plot of the (a) INL and (b) DNL versus output code at 500MS/s.

500 MHz system clock; this clock source provides  $< 5$  ps rms jitter. The dynamic performance at 500 MS/s is shown in Fig. 3-13(a). The periodic behavior of the SNDR is due to the time-interleaving. At the peaks of the SNDR curve, the input frequency is aliased, with reference to the channel sampling rate, to low frequencies. The low ERBW is a result of insufficient settling time in the input sampling network. This is verified with the SNDR plot of Fig. 3-13(b) measured at 125 MS/s, where the ERBW exceeds Nyquist with much

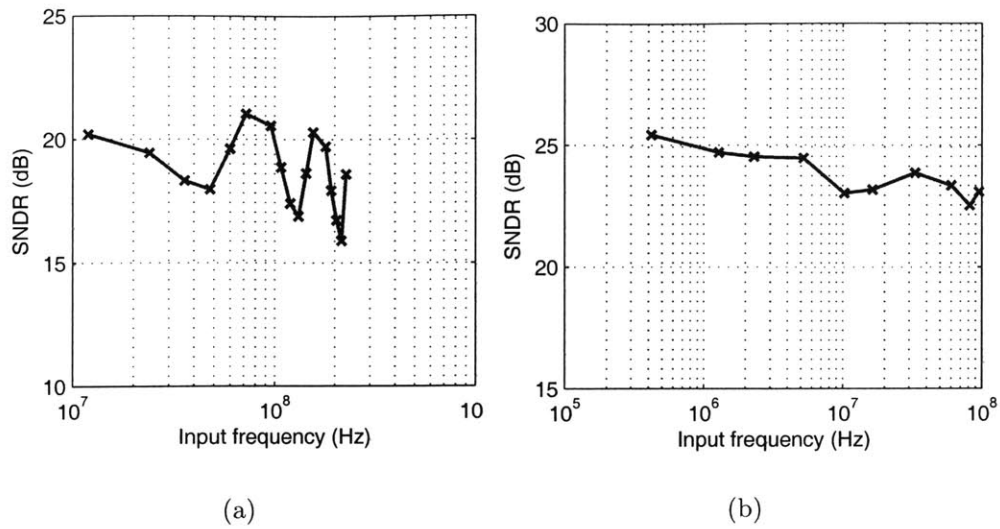


Figure 3-13: Plot of SNDR versus input frequency at (a) 500 MS/s and (b) 125 MS/s.

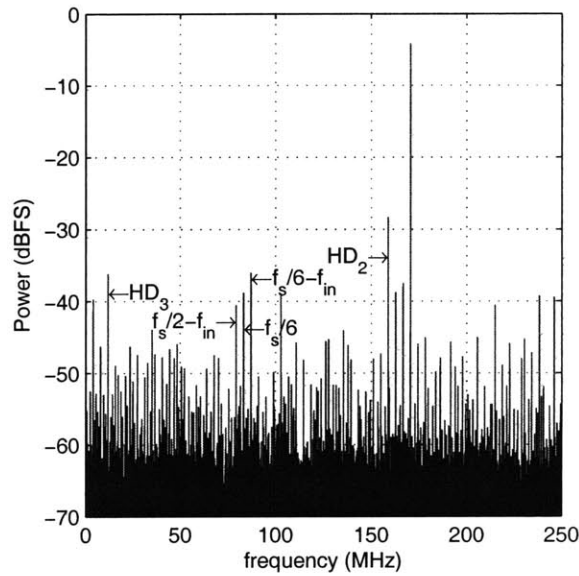


Figure 3-14: FFT of a 170.56 MHz input signal sampled at 500 MS/s.

less in-band variation. Besides the increased ERBW, the SNDR at DC also increases at the lower sampling rate. The reduction of SNDR at high sampling rates is possibly attributed to increased noise level from the digital output drivers and from insufficient settling time in the comparators. There is also an increased bit error rate (BER) [71] at 500 MS/s, indicating failures in the latch to fully resolve.

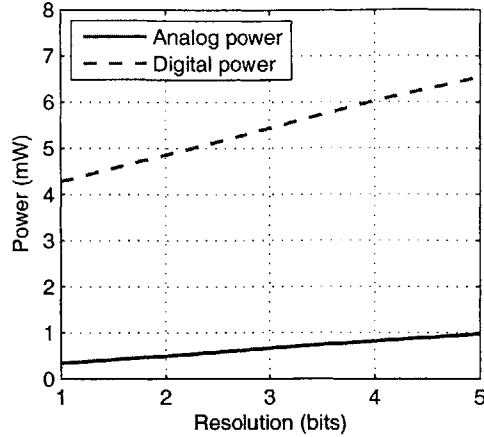


Figure 3-15: Analog and digital power consumption versus resolution for 500MS/s operation.

The FFT of a 170.56 MHz input sampled at 500 MS/s is shown in Fig. 3-14. The dominant harmonics are labeled. The spurs from offset between channels at multiples of  $f_s/6$  and from timing skew at  $f_s/6 \pm f_{in}$  are comparable to those of the third harmonic at -38 dBFS. The spurious free dynamic range (SFDR) is better than 24 dB, and the total harmonic distortion (THD) is better than -15 dB for input frequencies up to over 300 MHz.

Each ADC draws  $810 \mu A$  from a 1.2-V analog supply and 3.6 mA from a 1.8-V digital supply, for a total I/Q power consumption of 14.9 mW. Separating out the power for the comparators' latches (1.25 mW) from the total digital power, the logic consumes roughly 2.4 times the power of the comparators (the power drawn by the capacitor array is negligible). The energy model in Chapter 2 predicts that this ratio should be 3.7 (see Fig. 2-6). The difference is a result of optimization of the digital logic compared to the first model and the preamplifiers operating at a non-optimum bias point; as the ratio is lower than predicted, this further supports the original choice of SAR over flash.

The prototype includes an optional mode to disable the duty cycling of the preamplifiers. Duty cycling reduces the analog power by 15% at 500 MS/s and by 26% at 125 MS/s. At lower clock frequencies, the savings are greater because the overhead to charge and discharge the output capacitances of the preamplifiers is reduced in comparison to the energy savings over the longer time period that currents are disabled. Power dissipation versus resolution is

Table 3.1: Summary of measurements per ADC

Process	0.18 $\mu$ m, 2 poly, 5 metal CMOS	
Resolution	5 bits	
Active area	1 mm $\times$ 0.5 mm	
Input capacitance	640 fF	
Input range	800 mV <sub>pp</sub> (differential)	
Sampling rate	500 MS/s	125 MS/s
Analog power	0.97 mW @ 1.2 V	0.85 mW @ 1.2 V
Digital power	6.5 mW @ 1.8 V	0.82 mW @ 1.4 V
SNDR (DC)	20.2 dB	25.4 dB
SFDR	24 dB	39 dB
THD	-16 dB	-36 dB

shown in Fig. 3-15. As predicted, analog and digital power vary linearly with the resolution. The digital power does not asymptotically approach zero because bit scaling only reduces the number of bit decisions. The number of sampling operations remains constant, and sampling requires more energy than a bit decision. The measurements are summarized in Table 3.1.

The ADC has been successfully tested as part a complete UWB receiver capable of 100 Mb/s wireless transmission [5].



## Chapter 4

# Prototype Featuring Split Capacitor Array in Deep Sub-Micron CMOS

While the dynamic performance was not as strong as originally targeted, the first chip still achieved low overall power numbers. Looking at the results, the power is completely dominated by the switching power of digital circuitry, a problem that is usually not mentioned in the ADC literature (although see Chapter 5 to realize how common this problem is). In addition, while one of the primary challenges in porting analog designs to advanced technologies is designing for operation at the reduced voltage supplies, the analog supply voltage in the first testchip is at 1.2 V, already near or within the maximum voltage limits of deep sub-micron CMOS processes. The comparator achieves its gain through cascading and positive feedback, and it does not require cascoding or other techniques that increase the minimum supply voltage. Also, the performance of the passive capacitor array is determined primarily by matching and the speed of the switches. Thus, the SAR architecture, with its medium digital complexity, passive capacitor array, and comparator, seems better suited for implementation in more advanced technologies than an operational-amplifier based pipelined converter.

This chapter presents the design of a second ADC prototype, which is implemented in a low leakage 65-nm CMOS technology. Reduced feature sizes directly benefit the digital

logic controller, the block that limited the speed and power of the first prototype, but it introduces new challenges for analog circuitry, which suffer from reduced intrinsic gain and voltage headroom. This prototype implements a new capacitor array DAC structure, termed the split capacitor array, that maximally conserves charge during a conversion while exhibiting superior linearity than the binary weighted capacitive DAC without an increase in area. The self-timed bit-cycling used in the first chip is replaced with a delay-line based approach to adjust the latch strobing instant for increased preamplifier settling time in every bit decision, including the first. These features are described below, followed by measurement results demonstrating state-of-the-art performance.

## 4.1 A Foray into Charge Conservation: The SAR Capacitive DAC

The DAC serves two purposes in a SAR converter; it samples the input charge, and it generates an error voltage between the input and current digital estimate. The conventional DAC choice is a binary-weighted capacitor array [28], as shown in Fig. 4-1. Unlike the alternative serial DAC [86] or  $C-2C$  ladder [87] structures, the linearity of this binary weighted capacitor array is insensitive to parasitic capacitance between any of its nodes and ground. This array, however, uses charge inefficiently during a conversion. This section begins with an energy analysis of the conventional array structure and three different switching methods with successively improved energy efficiencies. Then, the split capacitor array is introduced, which is a new structure that retains all the benefits of the conventional array, but has reduced switching energy, improved linearity, and faster switching speed.

All the array structures share common functionality to implement the SAR algorithm. At the start of a conversion,  $S_{sample}$  is high, and the entire capacitor array stores the voltage  $V_{MID} - V_{IN}$ . Then, at time 0, the MSB cap,  $C_b$  is connected to  $V_{REF}$ , causing  $V_X$  to settle to

$$V_X[1] = -V_{IN} + \frac{V_{REF}}{2}. \quad (4.1)$$

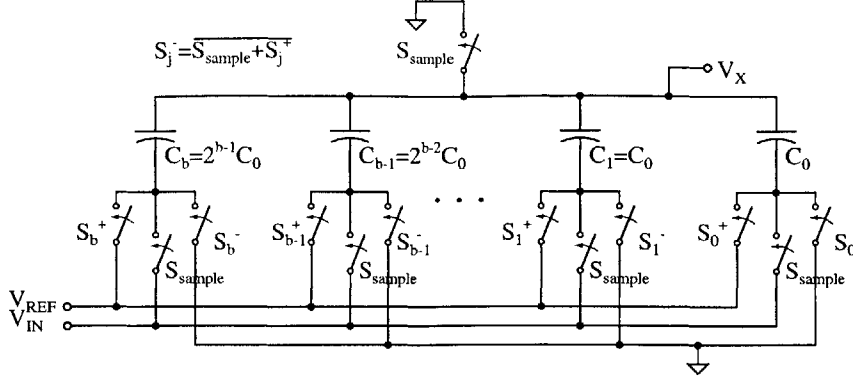


Figure 4-1: Conventional  $b$ -bit binary weighted capacitor array.

The comparator's output is then

$$D_1 = \begin{cases} 1 & V_{IN} < V_{REF}/2 \\ 0 & V_{IN} > V_{REF}/2 \end{cases} \quad (4.2)$$

The comparator's output controls the next switch transition. If  $D_1$  is low, the second largest capacitor is connected to  $V_{REF}$  ( $S_{b-1}^+ = 1$ ), raising the voltage at  $V_X$  (I will call this an "up" transition). If, on the other hand,  $D_1$  is high,  $C_b$  is returned to ground and  $C_{b-1}$  is connected to  $V_{REF}$  (a "down" transition).

The above process is repeated for successive capacitors in the array. At each stage, the value of  $V_X$ , after switching transients have settled, is

$$V_X = -V_{IN} + \frac{C_T}{C_T + C_B} V_{REF} \quad (4.3)$$

where  $C_T$  is the sum of all capacitors connected to the reference voltage, and  $C_B$  is the sum of all capacitors connected to ground:

$$C_T = \sum_i 2^{i-1} C_0 \quad \text{for } i \text{ such that } S_i^+ = 1 \quad (4.4)$$

$$C_B = \sum_i 2^{i-1} C_0 \quad \text{for } i \text{ such that } S_i^+ = 0 \quad (4.5)$$

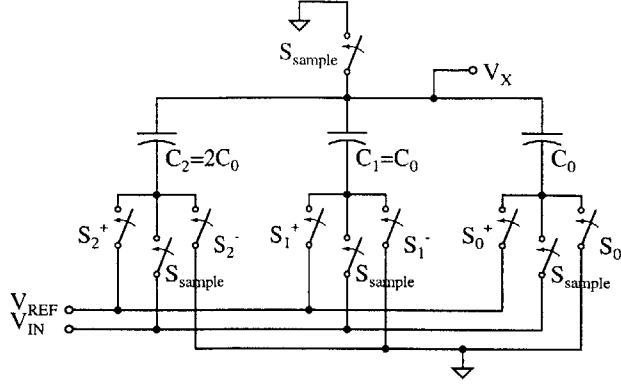


Figure 4-2: 2-bit capacitor array for 1 step and 2 step switching methods.

### 4.1.1 Capacitor Switching Methods

For ease of computation, all calculations in this section are done for the 2-bit capacitor array shown in Fig. 4-2. The equations presented can be generalized to an arbitrary size capacitor array, as is done in the Matlab model in Section 4.2.

Assume that at time  $0^-$ , the input voltage has been fully sampled on the capacitor array and that all switches are open. At time 0, the bottom plate of the  $C_2$  is switched to  $V_{REF}$ . The capacitor array is then charged to reach the final value in (4.1). If the capacitor array settles in time  $T_P$ , the total energy drawn from  $V_{REF}$  is

$$E_{0 \rightarrow 1} = \int_{0^+}^{T_P} i_{REF}(t) V_{REF} dt = V_{REF} \int_{0^+}^{T_P} i_{REF}(t) dt \quad (4.6)$$

Since  $i_{REF}(t) = -dQ_{C_2}/dt$  and charge continuity on a capacitor implies that  $Q_{C_2}(0^+) = Q_{C_2}(0^-) = 2C_0 V_X[0]$ , (4.6) simplifies to

$$\begin{aligned} E_{0 \rightarrow 1} &= -V_{REF} \int_{0^+}^{T_P} \frac{dQ_{C_2}}{dt} dt = -V_{REF} \int_{Q_{C_2}(0^+)}^{Q_{C_2}(T_P)} dQ_{C_2} \\ &= -V_{REF} (Q_{C_2}(T_P) - Q_{C_2}(0^+)) \\ &= -V_{REF} 2C_0 ((V_X(T_P) - V_{REF}) - V_X(0)) \\ &= C_0 V_{REF}^2 \end{aligned} \quad (4.7)$$

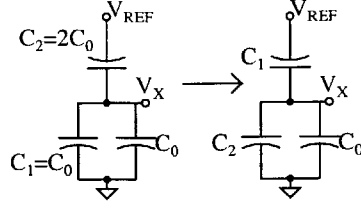


Figure 4-3: A “down” transition for the 1 step switching method.

This result matches the intuition that the effective series capacitance of the array,  $C_0$ , must be charged from 0 to  $V_{REF}$ . The notation  $V_X[i]$  will be used in the following calculations to indicate the value of (4.3) after all transients have died out for the  $i$ 'th transition, or, equivalently,  $V_X[i] = V_X(i \cdot T_P)$ .

At the end of the first bit decision period, the sign of the voltage at the top plate of the capacitor array is compared to  $V_{MID}$ , producing  $D_1$ . If  $D_1$  is 0,  $C_1$  in Fig. 4-2 is connected to  $V_{REF}$ , which is termed an “up” transition; all four switching methods behave identically in this case. The total energy drawn from  $V_{REF}$  for an “up” transition can be computed as follows:

$$V_X[2] = -V_{IN} + \frac{3}{4}V_{REF} \quad (4.8)$$

$$\begin{aligned} E_{1 \rightarrow 2} &= -V_{REF}[2C_0((V_X[2] - V_{REF}) - (V_X[1] - V_{REF})) \\ &\quad + C_0((V_X[2] - V_{REF}) - V_X[1])] \\ &= \frac{C_0 V_{REF}^2}{4} \end{aligned} \quad (4.9)$$

The following subsections present four different methods of performing the “down” transition if  $D_1$  is 1 instead of 0.

### 4.1.2 Conventional One Step Switching

For a “down” transition,  $C_2$  is switched down to ground and  $C_1$  is connected to  $V_{REF}$ . The most straightforward method to perform this transition is to switch the two capacitors simultaneously as shown in Fig. 4-3. The energy drawn from  $V_{REF}$  while the capacitor array

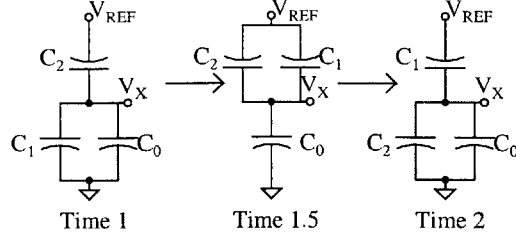


Figure 4-4: A “down” transition for the 2 step switching method.

settles can be determined by:

$$V_X[2] = -V_{IN} + \frac{V_{REF}}{4} \quad (4.10)$$

$$\begin{aligned} E_{1 \rightarrow 2, 1st\ step} &= -V_{REF}[C_0((V_X[2] - V_{REF}) - V_X[1])] \\ &= \frac{5}{4}C_0V_{REF}^2 \\ &= E_{TOP} + E_{SW} \end{aligned} \quad (4.11)$$

Here,  $E_{TOP}$  and  $E_{SW}$  represent the energy required to drive the change in  $V_X$  and to charge the new capacitor from ground to  $V_{REF}$ , respectively. They are defined as:

$$\begin{aligned} E_{TOP} &= -C_T V_{REF}(V_X[2] - V_X[1]) \\ &= C_T V_{REF}(-\Delta V_X) \end{aligned} \quad (4.12)$$

$$E_{SW} = C_0 V_{REF}^2 \quad (4.13)$$

While this method is the simplest to implement in terms of numbers of switches and clock edges, the ratio of (4.11) to (4.9) hints at a significant inefficiency because it requires five times more energy to lower  $V_X$  than to raise it by the same amount.

### 4.1.3 Two Step Switching

The second method, as used in the earliest charge redistribution SAR ADC [29], accomplishes the same transition using two switching steps instead of one. During the first step (considered from time 1 to 1.5), both  $C_2$  and  $C_1$  are connected to  $V_{REF}$ . This transition is identical to

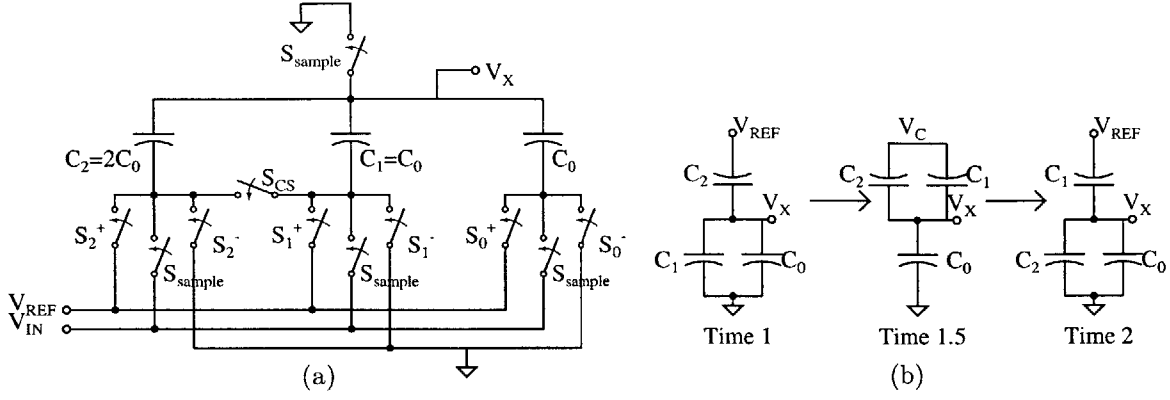


Figure 4-5: Capacitor (a) array and (b) equivalent circuits for the charge sharing method. When  $S_2^+ = S_1^+ = 0$  and  $S_{CS} = 1$ ,  $C_1$  is charged up from  $C_2$  with no energy drawn from the supply.

that in (4.8)–(4.9). Then, at time 1.5, the largest capacitor is disconnected from  $V_{REF}$  and connected to ground, as seen in Fig. 4-4, drawing energy

$$\begin{aligned}
 E_{1.5 \rightarrow 2} &= -C_0 V_{REF} (V_X[2] - V_X[1.5]) \\
 &= \frac{C_0 V_{REF}^2}{2}.
 \end{aligned} \tag{4.14}$$

The total switching energy is thus

$$\begin{aligned}
 E_{1 \rightarrow 2, 2\text{step}} &= E_{1 \rightarrow 1.5} + E_{1.5 \rightarrow 2} = \frac{3}{4} C_0 V_{REF}^2 \\
 &= E_{TOP} + E_{SW} + 2C_0 V_{REF} \Delta V_X
 \end{aligned} \tag{4.15}$$

Since  $V_X[2] < V_X[1]$ , the last term above is negative and (4.15) is less than (4.11). Qualitatively, some of the charge transferred to  $C_1$  is reused from  $C_2$ , similar to a charge recovery scheme in switching of large data buses [88].

#### 4.1.4 Charge Sharing (CS)

The previous subsection shows that energy savings can be achieved by using some of the charge from the largest capacitor to charge up the second capacitor. An extension of this

method can be used for additional savings. Fig. 4-5(a) shows a modified capacitor array, with an additional switch  $S_{CS}$ . As in the above subsection, the switching is done in two phases, as seen in Fig. 4-5(b). In the first phase, the first two capacitors are disconnected from  $V_{REF}$  and ground and connected to each other through the new switch. During this phase, no energy is drawn from  $V_{REF}$ . Using charge conservation, the voltage at node  $V_C$  at the end of the first phase is

$$V_C[1.5] = \frac{2}{3}V_{REF}$$

Therefore, the MSB/2 capacitor is effectively charged to two-thirds of the reference voltage with zero energy expenditure.

During the second phase, the MSB capacitor is connected to ground and the MSB/2 capacitor is connected to  $V_{REF}$ . The total energy dissipated is therefore

$$\begin{aligned} E_{1 \rightarrow 2, CS} &= -C_0 V_{REF} ((V_X[2] - V_{REF}) - (V_X[1] - V_C)) \\ &= \frac{7}{12} C_0 V_{REF}^2 = E_{TOP} + \frac{1}{3} E_{SW}. \end{aligned} \quad (4.16)$$

### 4.1.5 Split Capacitor Array

Even though charge sharing can save much of the energy in a “down” transition, some energy must still be spent charging up  $C_1$  to  $V_{REF}$  (namely, the total energy includes a contribution from  $E_{SW}$ ). To avoid charging any capacitor to  $V_{REF}$  during a “down” transition, the final method splits the MSB capacitor into two capacitors of value  $C_0$ , and then switches down one of them. This capacitor splitting results in the capacitor array shown in Fig. 4-6(a).

During the first bit decision,  $C_{2,1}$  and  $C_{2,0}$  are both connected to  $V_{REF}$ , dissipating the energy (4.7). After time 1, instead of connecting  $C_1$  to  $V_{REF}$ ,  $C_{2,1}$  is simply connected directly to ground, as in Fig. 4-6(b). This requires the energy

$$\begin{aligned} E_{1 \rightarrow 2, split} &= -V_{REF} C_0 (V_X[2] - V_X[1]) \\ &= \frac{1}{4} C_0 V_{REF}^2 = E_{TOP} \end{aligned} \quad (4.17)$$



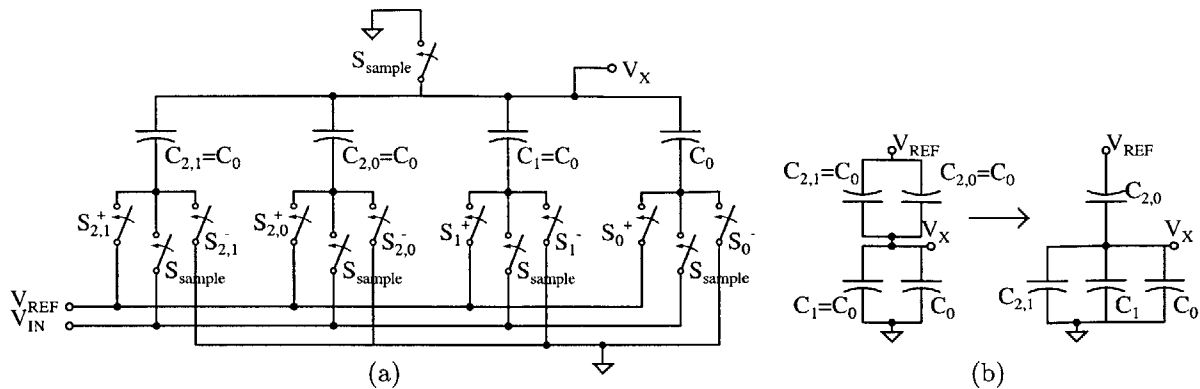


Figure 4-6: The (a) modified array and (b) switching method for a two-bit capacitor array where  $C_2$  has been split into two sub-capacitors,  $C_{2,1}$  and  $C_{2,0}$ .

Thus, the capacitor splitting approach requires no energy spent to charge up a capacitor from ground to  $V_{REF}$  during a “down” transition and has achieved the same energy for an “up” and a “down” transition.

The generalized  $b$ -bit split capacitor array is shown in Fig. 4-7; the MSB capacitor of the conventional array has been split into an identical copy (MSB subarray) of the rest of the array (main subarray). These arrays are placed in parallel (common top plate), not to be confused with the series connected capacitor arrays used in the sub-DAC approach.<sup>1</sup> The total capacitance of the split capacitor array is  $2^b C_0$  and identical to the conventional array. Because the capacitor array is usually arranged using a common centroid layout of multiple unit capacitors, no extra area is required. This approach requires twice as many switches; however, switches are generally small, so the area penalty is minimal.

The complete switching procedure is shown in Fig. 4-8. During the first bit cycle, all capacitors  $C_{b,0} \dots C_{b,b-1}$  in the MSB subarray are connected to  $V_{REF}$ . For subsequent “up” transitions, the  $C_i$  in the main subarray is connected to  $V_{REF}$ , while for any “down” transition, the MSB subarray capacitor  $C_{b,i}$  is connected to ground.

<sup>1</sup>Historically, the combination of capacitive main- and sub-DACs had been called a “split array” [42], but this has not become common usage, and I have co-opted the term for the new structure.

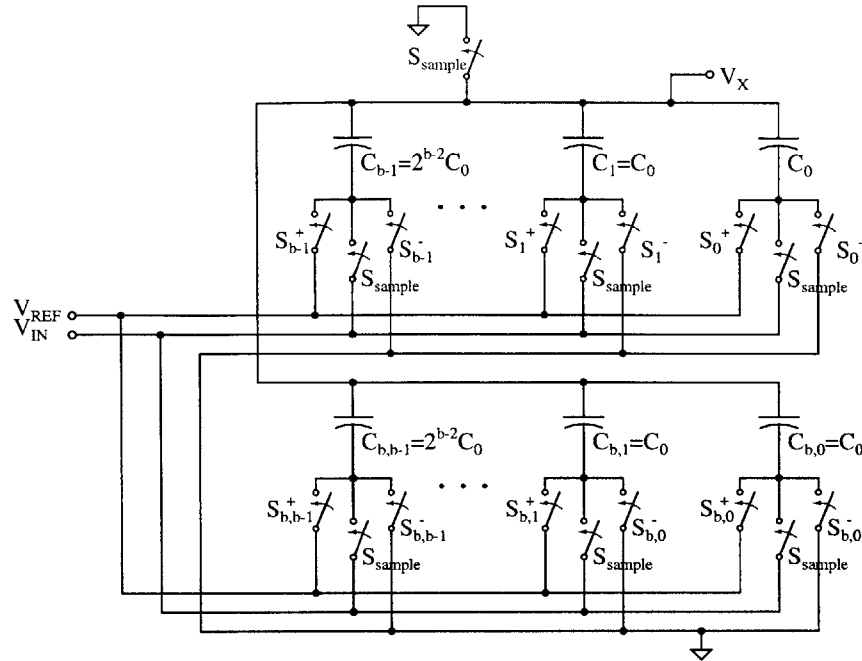


Figure 4-7:  $b$ -bit split capacitor array, with the main subarray on top and the MSB subarray below.

## 4.2 Energy Simulation Results

The above four switching methods have been applied to a 10-bit capacitor array. A Matlab model using generalized forms of (4.9), (4.11)-(4.13), and (4.15)-(4.17) has been developed. Fig. 4-9 compares this model to an HSPICE simulation of a 10-bit capacitor array in a CMOS 0.18  $\mu\text{m}$  process.

As can be seen in the figure, neglecting parasitics on the bottom plate of the capacitors causes the model to greatly underestimate the energy. The best capacitors available on the process are poly-poly capacitors with a 20fF parasitic bottom plate capacitance for  $C_0 = 100\text{fF}$ .

The simulations show clearly the energy savings that can be achieved using the different methodologies. At the highest output code, all transitions are “up,” so all of the switching methodologies require the same energy. At lower output codes, more “down” transitions occur in the array, and the split capacitor array exhibits its greatest energy savings.

The average overall switching energy for a uniformly distributed input signal as well as

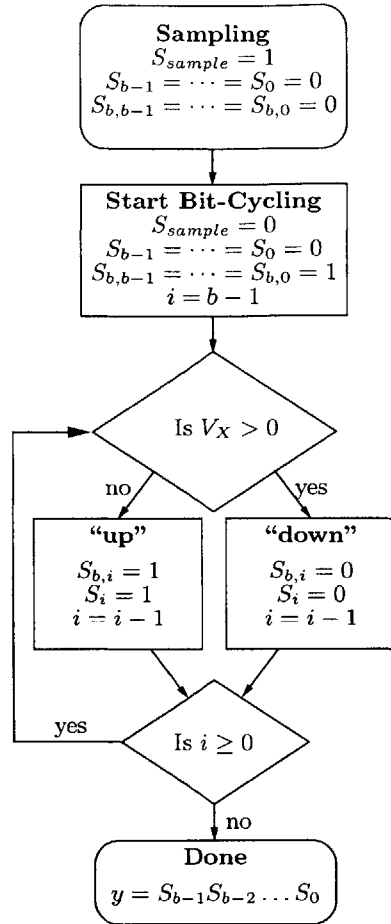


Figure 4-8: Switching procedure for split capacitor array.  $i$  represents the bit currently being decided.

for a full swing sinusoidal input can be calculated from the simulated data. The capacitor splitting method saves 37% of the switching energy compared to the conventional one step switching scheme. A comparison of the four switching methods is presented in Table 4.1.

The split capacitor array also draws current more evenly from a reference voltage buffer, if used. Both the overall current requirement is reduced, and any signal-dependent even-order nonlinearities of this current draw are eliminated.

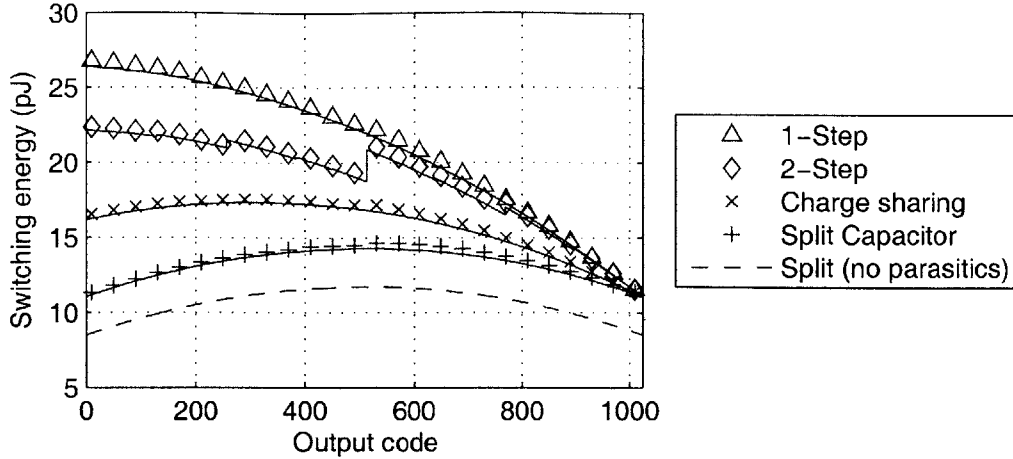


Figure 4-9: Plot showing the energy versus output code required for the switching of the capacitor array. The HSPICE data points are plotted against the Matlab models (solid lines). The bottom curve shows the effect of parasitics on the Matlab model.

Table 4.1: Switching methodology energy comparison

Method	$E_{UP}$	$E_{DOWN}$	# of Switches	# of Phases	$E_{uniform}$	$E_{sine}$
1 step	$E_{TOP} + E_{SW}$	$E_{TOP} + E_{SW}$	$b + 1$	1	1.000	1.000
2 step	$E_{TOP} + E_{SW}$	$E_{TOP} + E_{SW} - 2^i C_0 V_{REF} \Delta V_X$	$b + 1$	2	0.897	0.893
CS	$E_{TOP} + E_{SW}$	$E_{TOP} + E_{SW}/3$	$2b$	2	0.760	0.749
Splitting	$E_{TOP} + E_{SW}$	$E_{TOP}$	$2b$	1	0.635	0.623

#### 4.2.1 Switching Speed

For this high-speed implementation, an additional advantage of considerable significance is related to the array's settling time. During a "down" transition, two capacitors are required to switch for the conventional capacitor array; any mismatch, whether random or deterministic, in the digital logic driving these switches can cause the capacitor array to initially transition in the wrong direction, potentially exacerbating an overdrive condition for the preamplifiers. Only one capacitor in the split capacitor array transitions during any bit decision, providing inherent immunity to the skew of the switch signals. Simulation results comparing the settling times of the two arrays are shown in Fig. 4-10. For the simulations, the total width of the switches is identical for the split and conventional arrays. The split

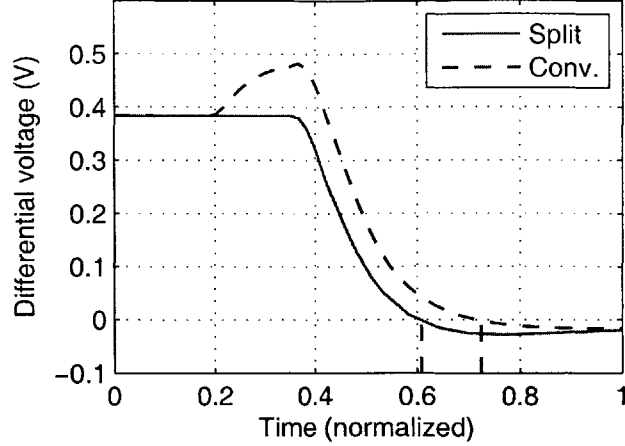


Figure 4-10: Simulation of the settling time of the split and conventional capacitor arrays under the presence of digital timing skew.

capacitor array settles up to 10% faster, which is used to reduce the bias currents in the preamplifiers by a similar amount.

## 4.2.2 Linearity Performance

To compare the theoretical static linearity of the binary-weighted and split DACs, each of the capacitors is modeled as the sum of the nominal capacitance value and some error term:

$$\begin{aligned}
 C_n &= 2^{n-1}C_0 + \delta_n \\
 C_{b,n} &= 2^{n-1}C_0 + \delta_{b,n}.
 \end{aligned}
 \tag{4.18}$$

Initially, consider only the case where all the errors are in the unit capacitors, whose values are independent identically-distributed (i.i.d.) Gaussian random variables; later in this section, other non-idealities will be considered. Then the error terms  $\delta_n$  and  $\delta_{b,n}$  have zero mean, are independent, and have variance

$$\text{E} [\delta_n^2] = \text{E} [\delta_{b,n}^2] = 2^{n-1}\sigma_0^2,
 \tag{4.19}$$

where  $\sigma_0$  is the standard deviation of the unit capacitor.

The linearity of a SAR ADC is limited by the accuracy of the DAC outputs, which are calculated here for the case of no initial charge on the array ( $V_{IN} = 0$ ). For a given DAC digital input  $y = \sum_{n=1}^b S_n 2^{n-1}$ , with  $S_n$  equals 0 or 1 represents the ADC decision for bit  $n$ , the analog output for the conventional binary-weighted array is

$$V_{X,conv}(y) = \frac{\sum_{n=1}^b (2^{n-1}C_0 + \delta_n) S_n}{2^b C_0 + \Delta C} V_{REF}. \quad (4.20)$$

The second term in the denominator  $\Delta C = \sum_{n=0}^b \delta_n$  will be neglected for this discussion. This will make the analysis simpler but will prevent a complete closed form solution for the integral nonlinearity (INL); see Appendix B for an analysis without this approximation. Subtracting the nominal value yields the error term

$$V_{err}(y) \approx \frac{\sum_{n=1}^b \delta_n S_n}{2^b C_0} V_{REF} \quad (4.21)$$

with variance

$$\begin{aligned} \mathbb{E} [V_{err}^2(y)] &= \frac{\sum_{n=1}^b 2^{n-1} \sigma_0^2 S_n}{2^{2b} C_0^2} V_{REF}^2 \\ &= \frac{y}{2^{2b}} \frac{\sigma_0^2}{C_0^2} V_{REF}^2. \end{aligned} \quad (4.22)$$

The voltage error in (4.21) and (4.22) is the sum of the errors from  $y$  unit capacitors connected to  $V_{REF}$ . Because the errors in the unit capacitors are assumed to be i.i.d., it does not matter which unit capacitors are connected to  $V_{REF}$  but only the total number. Thus (4.22) holds for the case of the split capacitor array as well. This error is also directly related to the INL of the ADC, and thus there should be no difference between the maximum INLs of the two arrays.

The differential nonlinearity (DNL) of the capacitive DAC is, neglecting gain errors, the difference between the voltage errors at two consecutive DAC outputs, as in

$$\text{DNL}(y) \approx \Delta V_{err}(y) = V_{err}(y) - V_{err}(y - 1). \quad (4.23)$$

The worst case DNL for the binary weighted capacitor array is expected to occur at the step below the MSB transition, where its variance is

$$\begin{aligned} \text{E} [\Delta V_{err}^2 (2^{b-1})] &= \text{E} \left[ \left( \frac{\delta_b - \sum_{n=1}^{b-1} \delta_n}{2^b C_0} V_{REF} \right)^2 \right] \\ &\approx \frac{\sigma_0^2}{2^b C_0^2} V_{REF}^2. \end{aligned} \quad (4.24)$$

For the split capacitor array, the worst case DNL also occurs at the step below the MSB transition, but its value is

$$\begin{aligned} \Delta V_{err} (2^{b-1}) &= \frac{\sum_{n=0}^{b-1} \delta_{b,n} - \left( \sum_{n=0}^{b-2} \delta_{b,n} + \sum_{n=1}^{b-2} \delta_n \right)}{2^b C_0} V_{REF} \\ &= \frac{\delta_{b,b-1} - \sum_{n=1}^{b-2} \delta_n}{2^b C_0} V_{REF}. \end{aligned} \quad (4.25)$$

This error has a variance of

$$\text{E} [\Delta V_{err}^2 (2^{b-1})] \approx \frac{1}{2} \frac{\sigma_0^2}{2^b C_0^2} V_{REF}^2. \quad (4.26)$$

Comparing (4.24) and (4.26) shows that the standard deviation of the worst-case DNL is  $\sqrt{2}$  lower for the split capacitor array. Conceptually, this occurs because the errors at  $y = 2^{b-1}$  and  $y = 2^{b-1} - 1$  are partially correlated for the split capacitor array, causing the cancellation of  $\delta_{b,0}, \dots, \delta_{b,b-2}$  in (4.25). This can be also be seen in the energy example above. In Fig. 4-3, the errors of the top capacitors are completely uncorrelated for the two bit decisions; however, in Fig. 4-6(b), the error of  $C_{2,0}$  contributes equally to both bit decisions.

A behavioral simulation of the SAR ADC, with both the binary weighted and split capacitor arrays, was performed. The values of the unit capacitors are taken to be Gaussian random variables with standard deviation of 3% ( $\sigma_0/C_0 = 0.03$ ), and the ADC is otherwise ideal. Fig. 4-11 shows the results of 10000 Monte Carlo runs, where the standard deviation of the INL and DNL are plotted versus output code at the 5-bit level. As expected, the conventional and split arrays have identical INL characteristics, and the split capacitor array

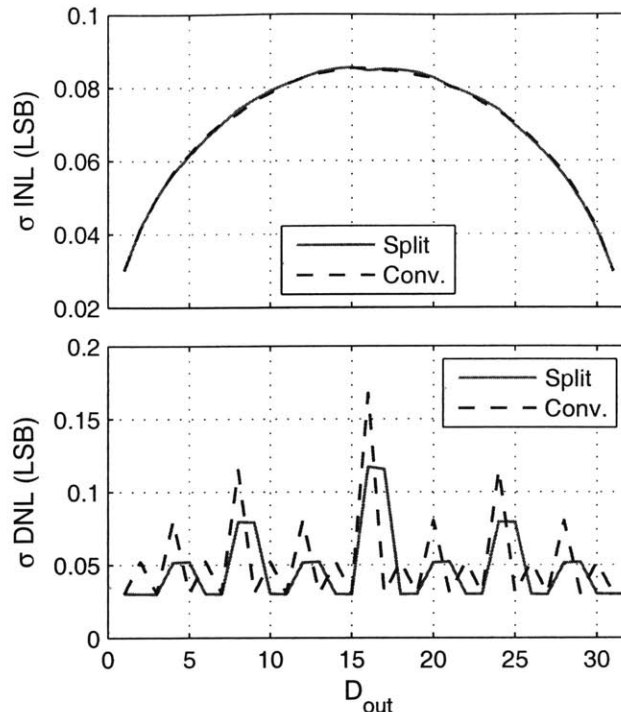


Figure 4-11: Behavioral simulation comparing the linearity of the split and conventional capacitor arrays. 10000 Monte Carlo runs were performed, with i.i.d. Gaussian errors in the unit capacitors ( $\sigma_0/C_0 = 3\%$ ). The standard deviation of the INL and DNL are plotted.

has  $\sqrt{2}$  better DNL. This improvement in DNL is similar to that conferred at the MSB transition from using 1-bit of unary decoding in a segmented DAC [89].

The above discussion assumes that the errors in the unit capacitors are due to an i.i.d. random process. In practice, care must be taken during layout to ensure absence of systematic non-idealities. The unit capacitors are arranged in a common centroid configuration to eliminate the effect of first order gradients. Fringing effects at the edge of the array are reduced by using 32 dummy capacitors around the 32 active unit capacitors. The largest capacitors in the main subarray and MSB subarray are distributed so as to have equal numbers of edges next to the dummy capacitors to further reduce fringing errors. The split capacitor array does have twice as many bottom plate signals that must be routed within the array. Coupling from these routes to the top plate routing can cause linearity errors and was avoided by routing the top and bottom plate signals distant from each other, which was



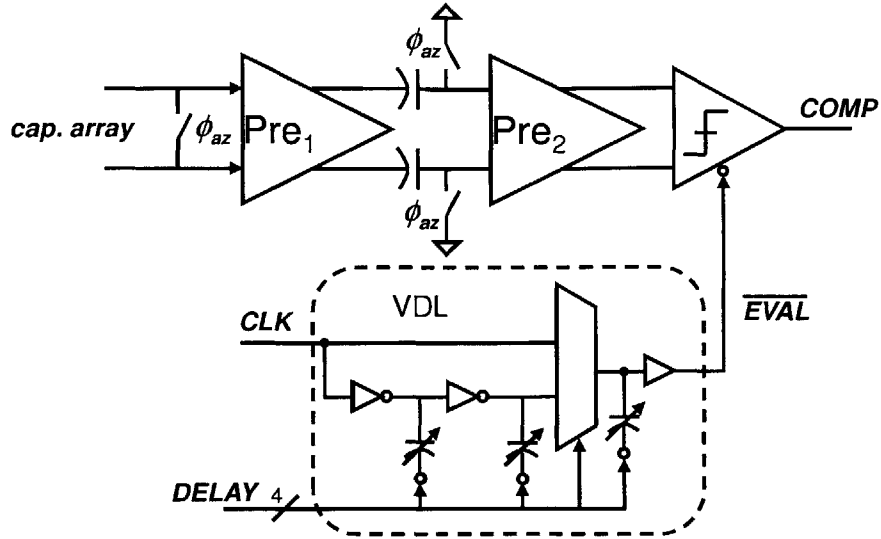


Figure 4-12: Comparator schematic showing preamplifier chain, latch, and VDL inserted in series with the latch strobe signal.

sufficient at 5-bit resolution. For higher resolutions, electrostatic shielding may be necessary where the bottom plate routing is separated from the capacitors by grounded metal [90]. Shielding can also improve immunity to noise coupling from the substrate.

### 4.2.3 Comparator With Adjustable Strobing

For each bit decision, the clock period is divided into one phase for the settling of the DAC and preamplifiers and one phase for regeneration of the latch. The latch typically resolves, even for small inputs, in much less than the 1 ns that is allocated assuming an even division of the period. The ADC sits idle after the latch settles until the start of the next bit decision period. Self-timed bit-cycling [37], as implemented in the first testchip, uses this idle time to start the next bit decision early. This approach relaxes the DAC and preamplifier settling time requirements for all but the decision determining the MSB, as it has no prior bit-cycle from which to borrow. Instead, here a variable delay line (VDL) has been inserted in series with the latch strobe signal to extend analog settling time in the first half of every bit decision, including the first, “pre-borrowing” time from that bit decision’s own latch phase. The beginning of every bit period is synchronous with the sampling clock, and the



14 plots the probability of a setup time violation versus the delay step for two differential input voltages. As expected, for a larger input voltage or shorter delay step, fewer setup time violations occur. The reduced error at delay step 10 is due to non-monotonicity in the VDL at this step. While the control loop's function is simple enough to implement on chip with a counter and simple finite state machine, this circuit exhibits a very high sensitivity to the input referred offset voltage of the replica latch. For a fixed differential input voltage, chosen to be zero for convenience, the delay of the replica latch to reach an output differential voltage of  $V_{latch}$  is [91]

$$t_{dlatch} \approx t_o + \tau \ln \left[ k \frac{V_{latch}}{V_{OS}} \right]. \quad (4.27)$$

Here,  $\tau = C/G_m$  is the time constant of regeneration, and  $k$  and  $t_o$  are topology and process dependent. As  $V_{OS}$  approaches zero, the second term in (4.27) increases without bound. Thus, this replica delay circuit can only be used to set the delay of the in-channel latches if the differential input voltage can be adjusted. An alternative approach is to directly embed the latch delay detection circuit within each channel. Busy input signals would exercise the entire input range of the latch, and actual latch errors can be detected and the delay step backed off accordingly. The delay calibration could be done continuously or shut down once a sufficiently low error rate is achieved. In the implemented prototype, the replica delay detection circuit was used, with its differential input controlled manually.

#### 4.2.4 Technology Considerations

In 65-nm CMOS, The SAR architecture's digital complexity directly benefits from the reduced feature sizes. Even though this ADC uses a fully static CMOS logic style, it still consumes less power than the highly customized logic, including dynamic registers, used in the first testchip. Care was taken throughout the digital logic to provide the maximum robustness in presence of delay variations.

The two analog blocks are well suited for integration in deep sub-micron CMOS with the following design considerations. For the same absolute device size, transistor matching improves in successive technology generations, allowing smaller total device area and capac-

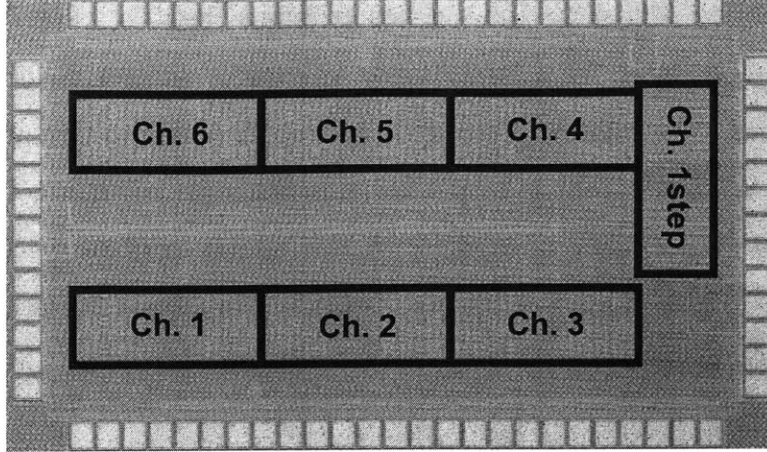


Figure 4-15: Photograph of  $1.9 \times 1.4 \text{ mm}^2$  die.

itance in the comparators [92]; however, the matching is not improved for minimum size devices. Also, due to the reduced power supplies and decreased  $g_m r_o$  of the short channel devices, it is difficult to get high gain in a single analog stage. The preamplifiers and latch use non-minimum length transistors to improve both the matching and output impedance. While this does increase device capacitance for the same  $g_m$ , there is only a small power impact because the extracted wiring parasitics, which do not scale with the gate length, are 65% of the total load capacitance for the first preamplifier.

The capacitor array is entirely passive, and its switching speed is improved with the shorter gate lengths. Because no analog-specific processing steps (e.g., a thin oxide for high density MiM capacitors) were used in fabrication the capacitors are formed using interdigitated metal comb capacitors. The capacitance is determined by fringing between adjacent metal lines, structures that have been shown to achieve similar densities to MiM capacitors with matching limits at greater than the 7-bit level [93]. The capacitance size is chosen according to the matching requirements discussed in Section 4.1. The input voltage is constrained to between 0 and 400 mV to allow sampling with a single standard- $V_T$  NFET transistor without exceeding the process voltage limit of 1.2 V.

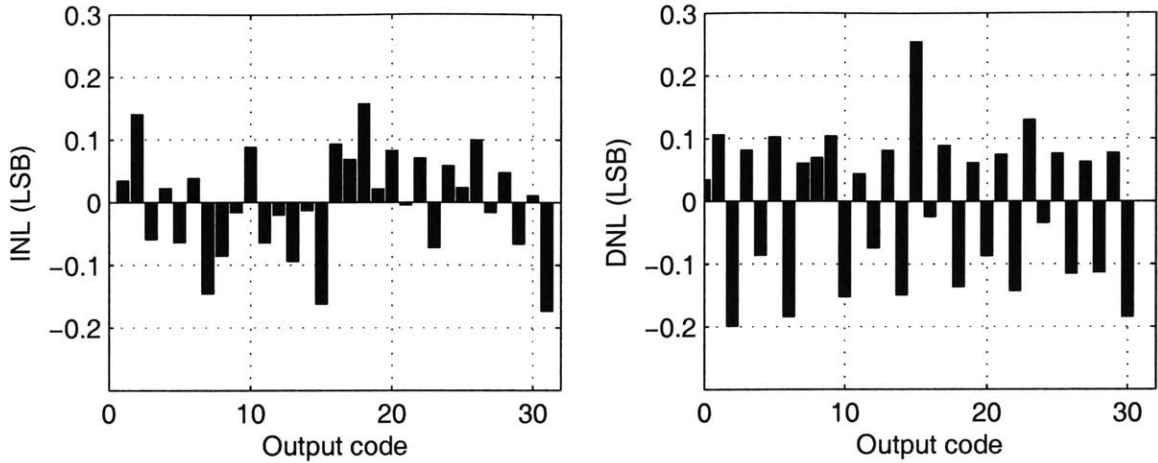


Figure 4-16: Static linearity of ADC versus output code.

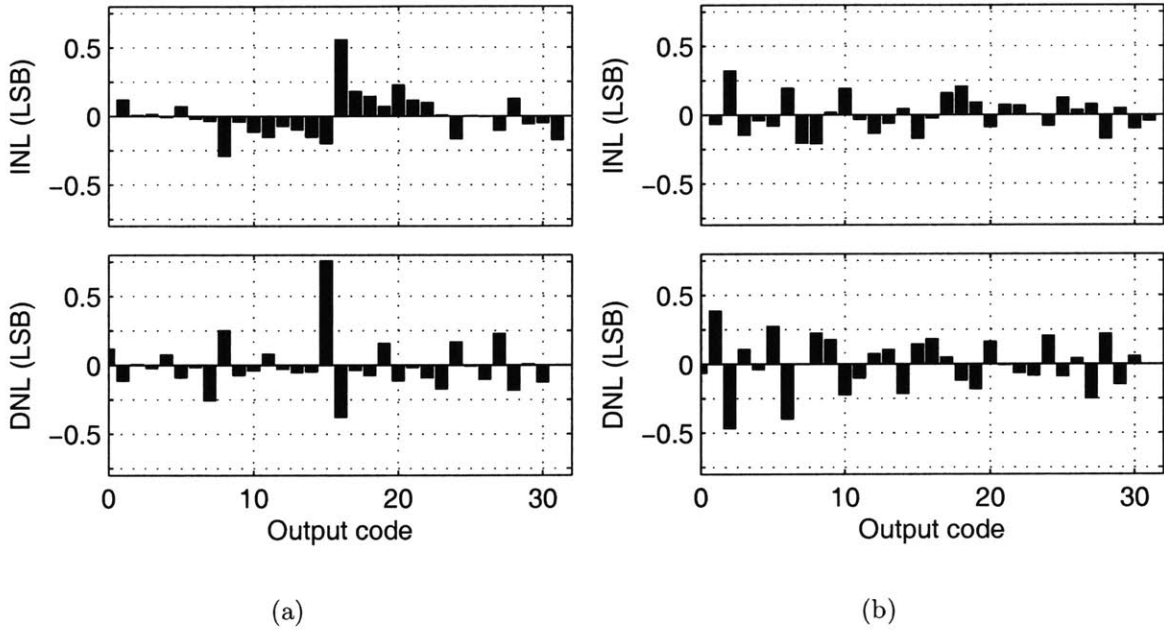


Figure 4-17: Comparison of the static linearity for the (a) conventional capacitor array and (b) the worst case split capacitor array channel on the same die.

### 4.3 Measurements

The ADC has been fabricated in a 65-nm CMOS technology; a die photograph is shown in Fig. 4-15. With a 91-kHz input sampled at 500 MS/s, the INL and DNL are -0.16/0.15 and -0.25/0.26 LSBs, respectively (Fig. 4-16). A separate test channel with the conventional

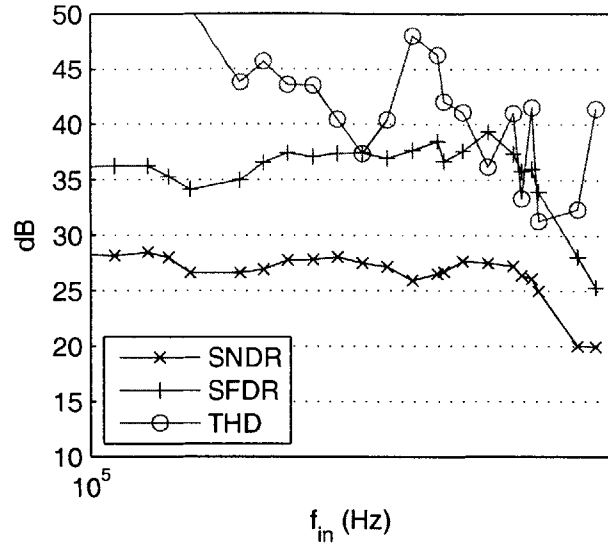


Figure 4-18: Dynamic performance versus input frequency.

capacitor array and one-step switching method is located on the upper right corner of the die. A direct comparison indicates that the split capacitor array uses 31% less power from the 400 mV reference voltage supply; the difference in energy savings from the theory presented above is due to the increased bottom-plate routing. The static linearity of the reference channel is plotted in Fig. 4-17(a) and has a peak DNL/INL of 0.76/0.56 LSBs. This is significantly higher than the overall ADC nonlinearity; however, the statistical measures used to calculate linearity [85] average any uncorrelated errors between channels, producing up to a  $\sqrt{6}$  improvement in static linearity for the interleaved results. A more suitable comparison is with a single channel. Figure 4-17(b) plots the static linearity of the worst case split capacitor channel on the same die as the conventional channel. While a direct comparison favors the split capacitor array, these results may be skewed by less diligent layout of the conventional array, leading to increased nonlinear coupling of the bottom plate to top plate routing.

The delay line was tested using the on-chip delay detection circuit and varying the input differential voltage, as shown in Fig. 4-14. Due to an underestimation of parasitics in the delay line, only the first 2 delay steps out of 16 provided sufficient time for latch regeneration, and these extended the period available to the preamplifiers by about 10%. At 250 MS/s, a

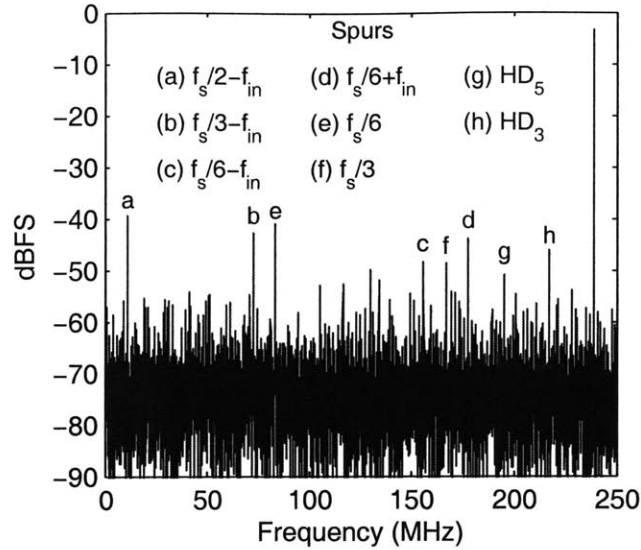


Figure 4-19: FFT of 239.04-MHz sine wave sampled at 500 MS/s; dominant spurs are labeled.

0.5-1 dB improvement in SNDR was achieved by properly tuning the delay.

The dynamic performance of the ADC is shown in Fig. 4-18 with the input frequency swept from DC to beyond Nyquist. The SNDR does not drop by 3 dB until past the Nyquist frequency. An FFT of a 239.04-MHz input is shown in Fig. 4-19. Spurs (a)-(d) result from gain errors and skew between channels, and spurs (e)-(f) are due to offset mismatch. All of these spurs are below -39 dBFS, and their combined power is still less than the total noise power (excluding the spurs) at this near-Nyquist input. The gain mismatch between channels is 0.9%. The individual channels have a low frequency effective number of bits (ENOB) between 4.65 and 4.75 that drops by 0.4 bits at Nyquist.

The ADC consumes 2.86 mW and 3.06 mW, respectively, from 1.2 V analog and digital supplies at the maximum sampling frequency. The ADC was also tested at lower sampling frequencies. At 250 MS/s, the ADC consumes a total of 1.58 mW from a 1 V digital and 0.8 V analog supply, while still maintaining Nyquist performance. A summary of the ADC is listed in Table 4.2.

Table 4.2: Performance summary of chip 2

Technology	65-nm CMOS 1P6M
Supply Voltage	1.2 V
Sampling Rate	500 MS/s
Resolution	5 bit
Input Range	800 mV <sub>pp</sub> Differential
SNDR ( $f_{in}=3.3$ MHz)	27.8 dB
SNDR ( $f_{in}=239$ MHz)	26.1 dB
SFDR ( $f_{in}=239$ MHz)	36.0 dB
THD ( $f_{in}=239$ MHz)	-41.5 dB
DNL (channel)	0.26 LSB
INL (channel)	0.16 LSB
Analog Power	2.86 mW
Digital Power	3.06 mW
Total Power	5.93 mW
Active Area	0.65 mm $\times$ 1.4 mm



# Chapter 5

## Mixed-Signal Optimum Energy Point

The two ADC prototypes presented previously demonstrated both the potential for high-speed SAR ADCs, and the energy-efficiency of this architecture. To further reduce the power, an optimization is performed to find the operating point with minimum energy for this interleaved SAR architecture. This chapter begins with a discussion of prior work in optimization of analog and digital circuits. It is noted that the class of ADCs with both significant analog *and* digital energy contributions cannot be optimized by focusing on the analog or digital components separately, but a systematic joint design and optimization of both parts and their interaction, must be performed. While optimization of the comparator energy was performed based on the equations in Section 2.1.3, and joint timing design of the analog and digital circuits was presented as part of the self-timing (Sec. 3.2.1) and VDL-based latch strobe point adjustment (Sec. 4.2.3), this chapter presents a far more comprehensive joint design of the analog and digital circuits. Timing is included in the model, including specific interactions and feedbacks (e.g., loading), and additional independent variables are added. In particular, the model is created to determine the optimum mixed-signal operating point, as specified by supply voltage and level of parallelism. This operating point achieves roughly  $3\times$  energy savings in the same technology as compared to the ADC presented in the previous chapter.

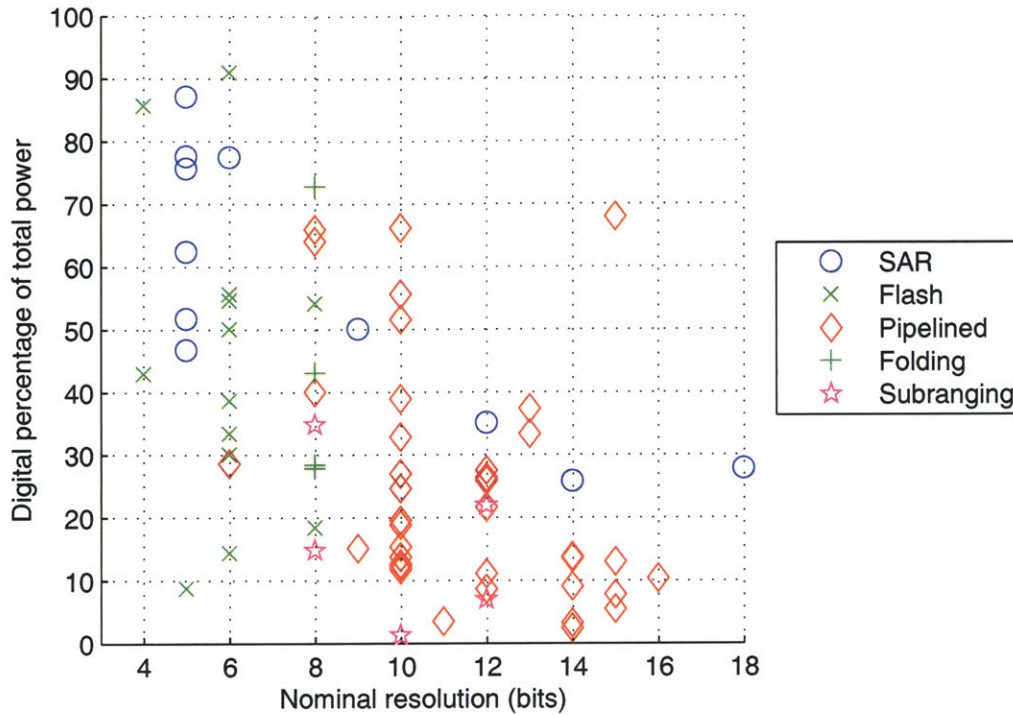


Figure 5-1: Relative digital power for published Nyquist ADCs that explicitly separate the analog and digital power consumptions.

## 5.1 Overview of Traditional Circuit Optimization

Over time, most ADC research has focused on architectures and techniques to reduce analog power. This has been encouraged by the scaling associated with Moore’s law that reduces digital power in comparison with analog and enables greater digital integration on chip. As a result of this trend, digital power is now a significant portion of the total dissipation in many ADCs, particularly those operating at low resolutions and high speeds. Figure 5-1 shows the percentage of digital power contribution in recent papers that differentiate between analog and digital power consumptions. While digital power can be significant across all the resolutions, it is particularly so at low resolutions and cannot be ignored, even though it is rarely mentioned in the literature.

SAR is an excellent architecture to demonstrate the advantage of joint analog and digital design because digital logic limits the critical feedback path, and at low-to-medium resolu-

tions, logic consumes a non-trivial portion of the total power, even at lower speeds (e.g., 34% at 12 bit, 100 kS/s [39]). The first two testchips used similar approaches for joint design, limiting it to timing adjustments. During every bit decision period, the ADC has two distinct phases: during the first phase, the capacitor array and preamplifiers (the linear section) are settling and the latch is in reset. During the second phase the linear section is presumed to have settled and the latch is strobed; it typically resolves very quickly, and the ADC would normally be idle after the latch has resolved. In the first prototype, self-timed bit-cycling [37] uses this idle time for the start of the next bit decision period, thereby hiding the latency of the digital feedback path and start of the capacitor array settling. The only bit decision that does not benefit from increased settling times is the first one because it is preceded by the sampling operation, which always ends at a fixed time. In measurements, it was the MSB decision that demonstrated the most errors at the highest sampling rates. A similar approach has been used to eliminate even the common strobe point of the latch, allowing fully asynchronous bit decisions with reduced logic and clock power [63].

There are many well known techniques for energy optimization in digital circuits. Perhaps the most straightforward way to save energy is to reduce the power supply voltage. The switching energy in digital circuits is proportional to  $V_{DD}^2$ , and so significant energy savings can result. The drawback is the decreased operating speed, which is inversely proportional to  $V_{DD}$ . Parallelism or pipelining can be used to maintain constant throughput at lower operating voltages at the expense of increased area and overhead in terms of multiplexing circuits or registers, respectively [94]. More complex algorithms use sensitivity functions to choose optimal device sizing [95] and select between different architectures [96].

Pure analog optimization exists as well, and much of it is in the context of ADCs. For instance, many papers have been published discussing optimum scaling between successive stages in pipelined converters [97–99]. A more comprehensive analog optimization for pipelined converters is presented in [100]. There, the specifications for the operational amplifiers, switches, and comparator are optimized for a combination of energy and area requirements using geometric programming, a type of convex optimization. Of these works,

only [97] addresses the digital circuitry, and only in a limited fashion. This chapter describes a framework and model for comprehensive analog/digital optimization.

One challenge of optimization not addressed by this thesis is the speed that the optimization can be solved. Discrete optimization is the broadest class of optimizations and the most straightforward to formulate. Any problem can be written as a discrete optimization. Unfortunately, without a lengthy exhaustive search of the design space, solutions may get trapped in local minima without ever visiting the global optimum, and failure to find a solution does not guarantee that one does not exist. Del mar Hershenson has published several papers describing setting up convex optimizations for analog circuitry and solving them using geometric programming. Formulations have been presented for operational amplifiers [101], inductors [102], LC-oscillators [103], phase-locked loops [104], amplifier chains [105], and pipelined ADCs [100]. The major challenge is formulating the problem as a convex optimization. Some constraints require significant manipulation to be made convex, and others simply cannot be included. In particular, only simple transistor models are amenable to convex formulations [101], and these models do not accurately characterize the behavior of advanced short-channel devices. Once set up, however, geometric programming led to solutions of complex design problems in much less than one minute, which would be almost instantaneous on a modern microprocessor. In comparison, the discrete optimization formulated in this chapter is solved in 20–30 minutes, making it less applicable for repeated design iterations, but it is able to handle a much wider range of problem definitions with a shorter setup time.

## 5.2 Modeling Methodology

This section describes the setup of the energy model and its simulation methodology. The speed of solving the energy model is not a primary concern for this work. Instead, the model emphasizes modularity for evaluating architectural changes and process independence. The process parameters can be calculated and entered into the architecture-based model after the fact. The formulated model ends up being a discrete optimization, the least efficient class of

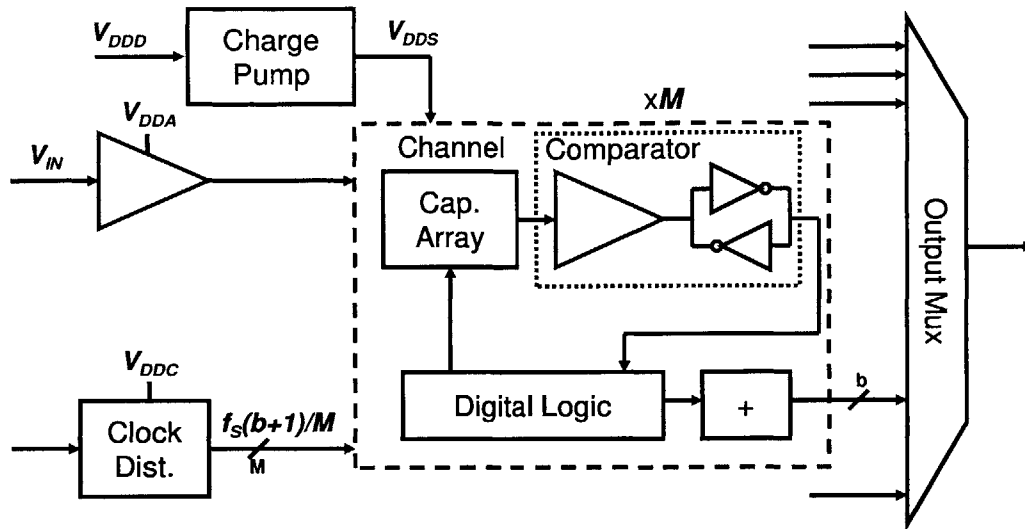


Figure 5-2: Blocks included in the comprehensive energy model. The blocks within the channel are replicated  $M$  times.

problem; however, it can still be solved for a local minimum within 20 minutes on a modern (circa 2006) microprocessor. Speeding up the solution and guaranteeing a global optimum are both worthy topics of research but fall beyond the scope of this thesis. The final model setup is specific to the SAR topology used in this thesis, but the framework of mixed-signal optimization described in this section is more generally applicable, and many of the blocks (e.g., preamplifier) and analog/digital interactions (e.g., sampling switch optimization) are useful in optimizations of a broader range of circuits and systems.

The ADC is initially partitioned into the set of hierarchical blocks, as shown in the conceptual block diagram of Fig. 5-2. The digital offset correction block is new to the channel (cf. Fig. 2-1(b)) and will be discussed in Section 5.3.4. The number of parallel channels,  $M$ , is now allowed to vary beyond 6. Increased parallelism requires additional complexity. Analog overhead includes the input buffer. The digital overhead includes an expanded clock network and an output mux to serialize all of the channels. Finally, a charge pump is introduced to allow independent setting of the core voltage from the sampling voltage, as described in Section 5.3.2.

Each block has a set of global inputs, local inputs, local variables, and outputs (Fig. 5-3). The global inputs are common to all blocks and are divided into a set of global optimization

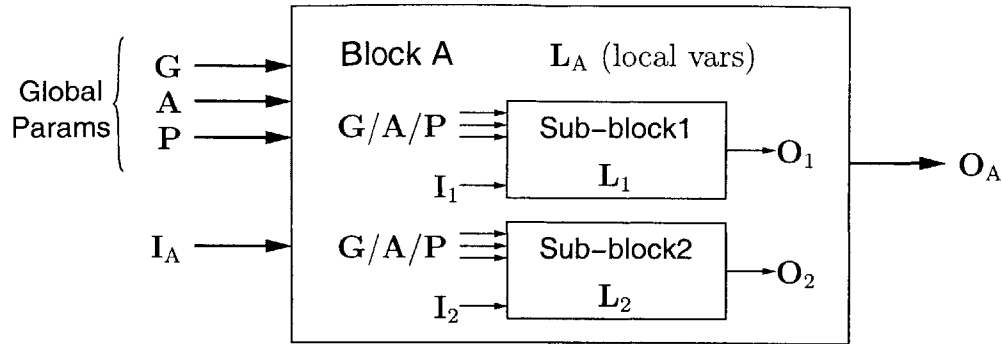


Figure 5-3: Basic block structure for the energy model. There are three sets of inputs common to all the blocks ( $\mathbf{G}$  = global optimization parameters (e.g.,  $V_{DD}$ ),  $\mathbf{A}$  = architectural parameters (e.g., timing partition variables), and  $\mathbf{P}$  = process-specific parameters (e.g.,  $g_m$ )). In addition, each block has its own specific inputs  $\mathbf{I}$ , local variables  $\mathbf{L}$ , and outputs  $\mathbf{O}$ . Any of these parameters may be passed to a sub-block contained within.

parameters, such as the voltage supplies, a set of architectural parameters that describe some switches in the block structure, and a set of process parameters. Thus, in a different technology, only the process parameters would have to change with an otherwise identical optimization setup. The line between the global optimization parameters and the global architectural parameters is arbitrary, as some of the global optimization parameters do require changes in the circuit topology, as in the sampling switch configuration described in Appendix C.2.1. A list of the global parameters is listed in Table 5.1, along with the default values and/or optimization ranges, as appropriate. These ranges are for the primary optimization that directly leads to the design in the following chapter. Different sets of global constraints are discussed at the end of this chapter.

The local inputs are determined by the enclosing block, or the top-level ADC if there is no enclosing block, and they are often outputs of another block. For instance, the energy required for the charge pump is determined by the total current drawn by the sampling switches in the digital block. Thus, the outputs of the blocks include the energy per conversion consumed as well as inputs for other blocks. This leads to the set of dependencies as shown in Fig. 5-4. The arrow  $\Rightarrow$  indicates that output parameters of one block are required to calculate the energy of another block. Because the actual number of connections between blocks (i.e., the circuit's signal path) is small, there are only a limited number of these model

Table 5.1: Global optimization parameters

Parameter	Description	Value/Range
$b$	Converter resolution	5
$f_S$	Sampling frequency	500 MHz
$M$	Number of time-interleaved channels	$\geq 1$
$V_{DDD}$	Core digital supply voltage	0.5–1.2 V
$V_{DDA}$	Core analog supply voltage	$V_{DDD}$
$V_{FS}$	Full scale input voltage	0.1 V– $V_{DDD}$
$V_{DDS}$	Boosted sampling voltage	$V_{DDD}$ –1.2 V
$SW_{SAMP}$	Sampling switch configuration (see App. C.2.1)	$\{N, TG\}$
$SW_{REF}$	Reference voltage switch configuration (see App. C.2.1)	$\{N, P\}$
$t_{conv}$	Total conversion time for a channel	$M/f_S$

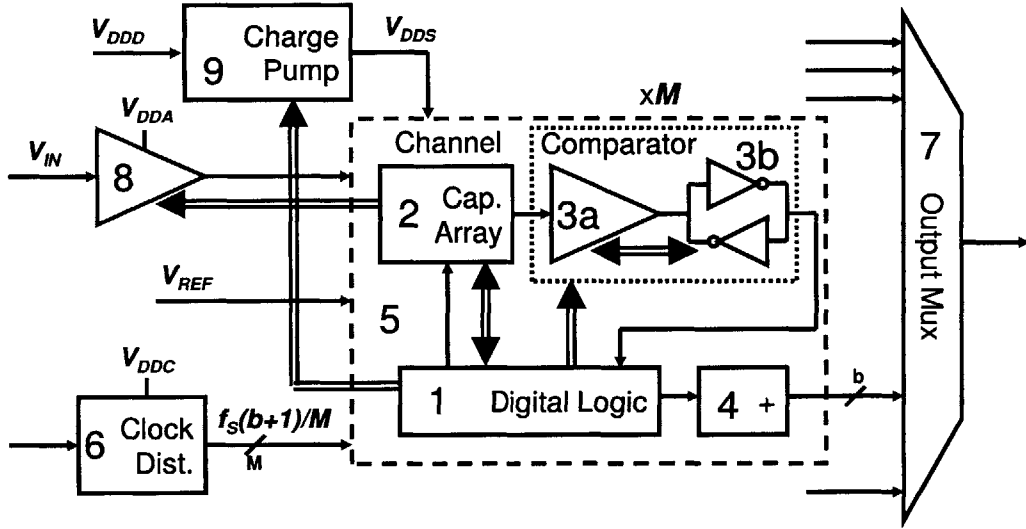


Figure 5-4: Data dependencies and solution order for the energy model blocks.

dependencies. The solution order (numbers in Fig. 5-4) is chosen such that each block’s energy is computed after all of its inputs are available. Two special cases can be seen where there are bidirectional dependencies, and these require special consideration in the model setup.

The first is within the comparator, where tradeoffs can be made between the size of the latch and the power of the preamplifiers. Specifically, a larger latch increases its input capacitance and switching energy but reduces the required preamplifier gain for a fixed input referred offset voltage. The second complicated dependency involves the digital buffers

driving the switches in the capacitor array, where additional buffering stages are inserted to drive the large fanout of the array. The added buffers decrease the time available for settling within the array, requiring an increased switch size and potentially more buffering. The methods to resolve these dependencies are discussed in the comparator and the channel block descriptions.

For a fixed set of optimization parameter values, the SAR energy model returns the energy per conversion, as well as the local variables that specify, among others, transistor sizing throughout the comparator and buffering required between the logic block and the capacitor array. There are some sets of optimization parameters, however, for which the model fails to find a solution, mostly because no solution is possible. For instance, the specification  $M = 1$  and  $V_{DD} = 0.5$  means a single 500 mV channel operating at 500 MS/s, requiring an internal clock rate of 3 GHz. As the delay through a single digital gate at 500 mV is much greater than 300 ps, no possible solution exists. The block descriptions below specify under what conditions failures can arise. A failure in any block will cause a failure of the entire SAR model. The optimization then searches the design space (Table 5.1) for the set of global inputs that yield a functioning design with the minimum modeled energy.

## 5.3 Block Descriptions

This section describes the blocks of the energy model. The order of the block descriptions is the same as the solution order in Fig. 5-4. Only the highlights and limitations of the block models are presented here; detailed descriptions and equations are included in Appendix C, along with a summary of the required process data.

### 5.3.1 Digital Logic

The energy of the digital logic within the channel is composed of the core energy necessary for the logical function of the successive approximation register plus the fanout energy necessary to drive the potentially large switches of the capacitor array. The former is fixed once



the resolution and supply voltage are set, and it is modeled identically to the model in Section 2.1.1 ( $E_{dig,core} \propto bV_{DD}^2$ ). The latter is dependent on and must be jointly optimized with the capacitor array sizing. As the capacitive load of the capacitor array switches can be large, buffering is automatically inserted to maintain a minimum size core controller with no gate having a fanout exceeding 4.

The model, detailed in Appendix C.1, includes the effect of leakage currents and the overhead for level shifting if a separate sampling supply voltage is used. In addition to the energy in (C.6), the digital block also calculates the setup and propagation delays in (C.8)–(C.10), which are required for the timing of the other blocks in the channel (see below). The delays include the effect of the buffer insertion.

The energy and delay of digital logic vary across process corners, and delay in particular can be significantly affected by local variations at lower voltages. The energy model only handles a single corner and is not variation aware. While the optimum energy point will change at different corners (e.g., more parallelism at higher digital energies), there will still be a broad minimum (cf. Fig. 5-9(a)) and therefore a wide range of reasonable levels of parallelism.

### 5.3.2 Capacitor Array

The capacitor array model from Section 2.1.2 is extended here to address the energy required to sample a high frequency signal and switch the array to meet a settling time constraint. The times available for sampling and switching are set by the channel block as described below, and they are considered fixed within the capacitor array model.

During the sampling operation, all of the capacitors in the array are connected in parallel, forming the basic sampling network shown in Fig. 5-5(a). The sampling network must meet two basic requirements. When the sampling switches first close, the charge across the capacitor must step from the previous stored value to the now, potentially different input voltage. The transient from this step response must settle within the sampling window, and this limits the maximum equivalent resistance of the two switches. Second, the network

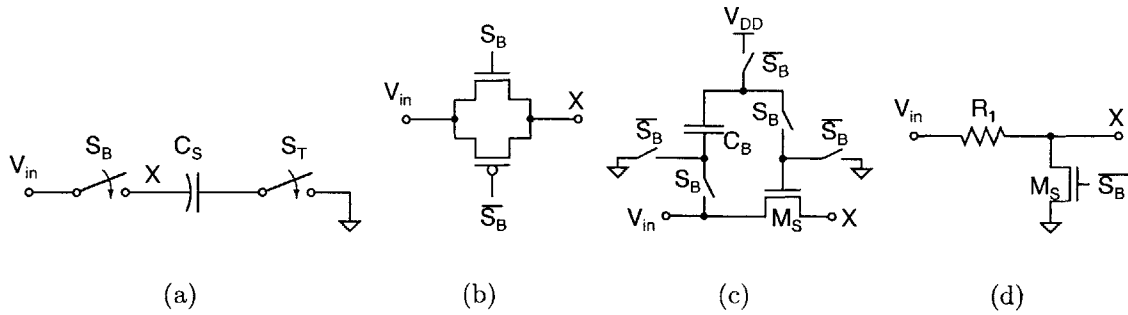


Figure 5-5: Bottom plate switch implementations for (a) the basic sampling network. Alternatives to the basic NFET bottom plate sampling switch include (b) transmission gate, (c) constant- $V_{GS}$  [106], and (d) switched- $RC$  [107] sampling switches.

must be able to track the highest frequency input signal without distortion. Distortion is primarily limited by the bottom plate switch  $S_B$ , the simplest implementation of which is a single MOSFET whose conductance is

$$g_{ds} \approx \frac{1}{\mu C_{ox} W/L (V_{gs} - V_T)} = \frac{1}{\mu C_{ox} W/L (V_{on} - V_{in} - V_T)}. \quad (5.1)$$

For a fixed on-gate voltage  $V_{on}$ , the conductance varies with the input voltage level, creating an input signal dependent (and nonlinear) attenuation and phase shift. This is particularly a problem in deep sub-micron CMOS with limited supply voltages because  $V_{DD} - V_T$  is relatively small, limiting the maximum  $V_{in}$  before the switch turns off. A transmission gate with both a PFET and NFET (Fig. 5-5(b)) extends the maximum input signal range, but if the supply voltage is less than the sum of the PFET and NFET threshold voltages ( $V_{Tp} + V_{Tn} > V_{DD}$ ), the transmission gate fails to turn on in the middle of the input signal range. The top plate switch,  $S_T$ , does not suffer from the same problem because its source voltage is always fixed at the DC reference point for the top plate, which can be set low to maximize the overdrive on this switch.

Several circuits have been proposed to accurately sample high frequency signals or sample with low voltages. Bootstrapping, or constant- $V_{GS}$  sampling, [108–110] sets  $V_{on}$  to track  $V_{in}$  giving a constant gate overdrive across the entire input signal range. Figure 5-5(c)

shows a simplified diagram of a bootstrapping network. Before sampling, during phase  $\overline{S_B}$ , the sampling switch  $M_S$  is turned off with its gate grounded. The boost capacitor  $C_B$  is precharged with a voltage  $V_{DD}$  across it. Then, during the sampling period,  $C_B$  is placed between the input voltage and the gate of the sampling switch, setting the latter to  $V_{DD} + V_{in}$ , which makes the  $V_{GS}$  of  $M_S$  independent of the input signal voltage. The backgate effect causes some residual input-signal conductance variation because of the dependence of  $V_T$  on  $V_{SB} = V_{in}$  if the bulk is grounded. If a triple well process is available or PFET sampling is used [111],  $V_{SB}$  can be set to zero, eliminating this residual error, and more sophisticated bootstrapping techniques have been proposed to adjust  $V_{on}$  to track  $V_{in} + V_T$  [112]. These bootstrapping techniques add complexity and power to the channel design, and care must be taken to avoid transients that can lead to voltages beyond the reliability limits of the process.

Several papers use closed-loop sampling networks and switched opamps [113, 114] to eliminate the floating series bottom plate switch, but these are not considered further here because the use of an operational amplifier violates one of the key principles motivating this thesis for design in deep sub-micron CMOS. Another approach, switched-RC sampling [107], eliminates the series bottom plate switch without requiring an opamp. The bottom plate switch is replaced with a series linear resistor and shunt switch, as shown in Fig. 5-5(d). During the track phase, the series resistance is linear, and during the hold phase, the shunt transistor  $M_S$  with a maximum overdrive voltage, strongly connects the bottom plate of  $C_S$  to ground, attenuating the input at node  $X$ . The linearity is limited only by the residual input signal at node  $X$  during the hold phase, but it leads to a DC current draw during the hold phase. For higher input frequencies,  $R_1$  must be small to sample accurately, but the hold current increases proportionately. Because of the very high maximum input frequencies, the switched-RC network is not appropriate for the ADC considered in this thesis.

To simplify the design and minimize complexity within the channel, this model only considers the simple NFET or transmission gate sampling networks but with an extra degree of freedom, which allows the voltage  $V_{DD_S}$  driving the NFET gate to be higher than the core

digital voltage  $V_{DDD}$ . This higher voltage can be generated using a charge pump, as discussed in Section 5.3.9.  $V_{DDS}$  is constrained to be within the process limit of 1.2 V, but this extra degree of freedom allows the core voltage to be driven lower and provides overall energy savings for the ADC (cf. Fig. 5-12). Appendix C.2.1 details the NFET sampling network. The transmission gate requires a more complicated model setup and is described in its own Appendix D. Appendix C.2.2 analyzes the switching of the array during the bit decision periods, which is similar to the sampling setup but with only the settling time constraint (i.e., no tracking requirements).

### 5.3.3 Comparator

The comparator model is designed to be more accurate and flexible than the model presented in Chapter 2. The topology uses a cascade of low gain preamplifiers, each with OOS, followed by a regenerative latch, but now the number of preamplifiers is variable. In addition, the tradeoff between latch sizing, offset, capacitance, and energy is made explicitly, and effects such as slewing and fixed wiring capacitances are included. Equations for the comparator, preamplifier, and latch are given in Appendix C.3.

The model does not include any effect of circuit noise, which is not significant at 5 bits, but it limits the applicability at higher resolutions. The direct effect of offset from local variation is considered for the latch and completely eliminated by OOS for the preamplifiers; however, mismatch can also reduce the effective gain of the preamplifiers below the model's output, particularly if the nominal gain is above 5.

### 5.3.4 Digital Offset Correction

Offset has two deleterious effects in interleaved ADCs. The first can be seen by considering the digital output of an ADC

$$y[n] = \mathcal{Q}(v_{IN}[n] + v_{OS}[n \bmod M]), \quad (5.2)$$

where  $Q(\cdot)$  represents the quantization function and  $v_{IN}[n]$  is the sampled analog input signal. For an interleaved ADC, the offset voltage  $v_{OS}$  is periodic with the number of channels, producing spurs in the output spectrum at multiples of  $f_S/M$ , the channel sampling frequency.

Offset produces a second effect, namely clipping of large input signals. The maximum input dynamic range of an ADC is thereby reduced by the peak-to-peak offsets across the channels. The first two testchips set the offset requirement of the comparator (i.e., the analog offset) to make the power of the periodic term in (5.2) smaller than the quantization noise power, namely

$$E[v_{OS}^2] = \sigma_{v_{OS}}^2 < \frac{V_{LSB}^2}{12}. \quad (5.3)$$

This condition forces the offset of every channel to be small ( $< V_{LSB}$ ), and the reduction in dynamic range due to clipping is negligible.

Another approach to reducing the power of the periodic term in (5.2) is to correct it digitally, which simply requires an adder<sup>1</sup> [50, 53]. Then, the analog offset requirements can be set only to keep clipping within an acceptable limit. The effect of digital offset correction is quantified using a coupling of the behavioral (see Appendix A) and energy models.

Figure 5-6 shows the SNDR of an interleaved ADC versus the analog offset voltage for varying degrees of digital offset correction. The lower plot, with no offset correction, tails off directly due to the effect of the periodic offset term in (5.2), while the upper curves, with digital offset correction, only drop when clipping becomes significant. For the same 1 dB degradation in SNDR, digitally correcting the offset to within a half an LSB permits using a 6× larger analog offset voltage, although 3× is used to be conservative in this design.

The results of the behavioral model can then be fed back to the energy model. Specifically,

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<sup>1</sup>While medium complexity calibration circuitry is required to determine the proper offset correction value, once calculated, the continual correction only requires an adder at the output of the channel. Section 6.3 further explores the contrast between calibration and correction.

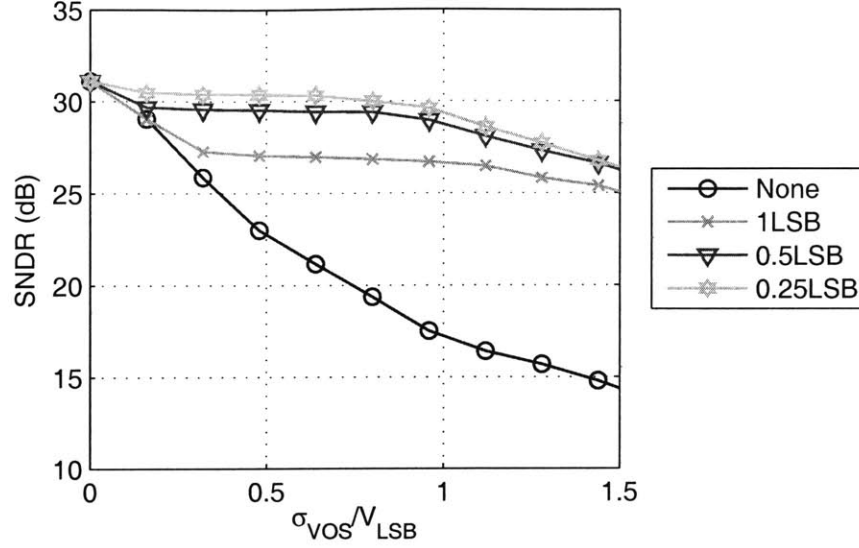


Figure 5-6: Behavioral simulations showing the performance degradation of the ADC versus the analog offset voltage for increasing degrees of digital offset correction. The ADC is otherwise ideal and is sampling a -1 dBFS sinusoidal input.

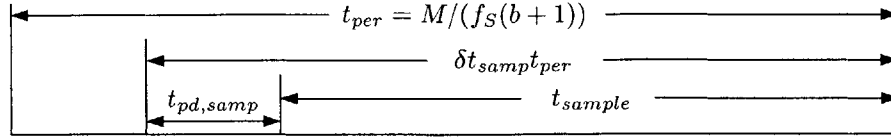
the comparator offset requirement in (C.32) is

$$V_{OS,in} = \begin{cases} \frac{2^{-b}V_{FS}}{6} & \text{no digital offset correction,} \\ \frac{2^{-b}V_{FS}}{2} & \text{with digital offset correction.} \end{cases} \quad (5.4)$$

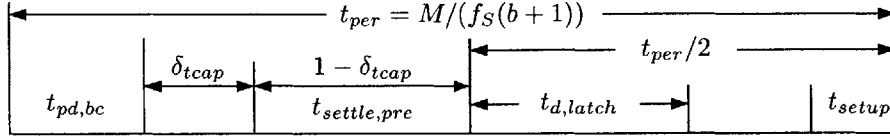
The increased comparator offset voltage permits a reduction of the number of preamplifiers from 2 to 1, and the energy model predicts an overall energy savings of 30% from using the digital offset correction. This demonstrates both the effectiveness of digital correction for analog impairments and the ability for the coupled behavioral and energy models to quantify architectural tradeoffs across the analog/digital boundary.

### 5.3.5 Channel

The channel block is responsible for setting up the timing of all the elements within the channel. There are two distinct operations performed by the channel, sampling and the bit decision, each with its own timing. The sampling period (Fig. 5-7(a)) is divided between



(a)



(b)

Figure 5-7: Timing division of the (a) sample and (b) bit decision clock periods.

the digital logic and the capacitor array, with the total sampling window set as a fraction,  $\delta t_{samp}$ , of a channel clock period ( $t_{conv}/(b+1)$ ), as specified in (5.5). This fractional term can be used to model architectural limitations to the length of the sampling window, such as avoiding overlapped sampling windows between the parallel channels.

$$t_{sample} = \frac{\delta t_{samp} t_{conv}}{b+1} - t_{pd,samp} \quad (5.5)$$

The bit decision clock periods are split into two half clock periods. The first half clock period consists of the propagation delay through the digital logic and the linear settling of the capacitor array and preamplifiers. The rest of the clock period is used for the regeneration time of the latch and the setup time of the digital logic, as shown in Fig. 5-7(b). The bit decision period does not need to be split evenly between these two phases; the VDL of Section 4.2.3 explicitly changes the timing between the two, which will be further examined in the context of the energy model in Section 5.4.3. For the purposes of this section and the final chip, the two phases will be of equal length. The linear analog settling time is split between the array and preamplifiers with architectural parameter  $\delta t_{cap}$ . This parameter can itself be determined by adding it to the global optimization setup (Table 5.1), and the value of 0.25 has been determined as optimal for the final results.

Appendix C.5 discusses the simulation setup to solve these timing constraints and focuses on the interplay between the timing of the digital and capacitor array blocks.

### 5.3.6 Clock Distribution

Clock distribution is one of the principle overheads of highly interleaved ADCs because both the number of clocks must increase and the die area across which the clocks are distributed is increased. For the proposed energy model, the clock distribution is implemented using a simple clock tree model. Generation (i.e., a PLL and dividers) is assumed to be handled at a higher level of the system hierarchy. To minimize timing skew, the clock paths to every channel should be balanced, which means that the paths to the closest and farthest channels should have the same length. This can be accomplished by an H-tree [115] or similar network. The length of the clock path to any channel therefore grows as the length from the center of the die (presumed to be the origin of clock distribution) to the corner of the die (the location of the farthest channel). If the die area grows with  $M$ , then the length of the clock path to a single channel grows with  $\sqrt{M}$ , making the per-channel clock capacitance

$$C_{clk} = C_{clk0} \sqrt{M/M_0}, \quad (5.6)$$

where measurements or parasitic extraction can be used to determine the capacitance  $C_{clk0}$  for distributing the clock to one out of  $M_0$  channels. This equation, simple in form, is very similar to the expected scaling of an H-tree's capacitance,  $C_{clkwire}$ , with the number of gates,  $N_g$ , in a digital circuit, which scales as  $C_{clkwire} \propto \sqrt{N_g}$  [116].

The total energy per conversion is

$$E_{clk} = C'_{clk} V_{DD}^2 (b + 1), \quad (5.7)$$

where  $C'_{clk}$  is modified to include the energy required for buffering as in (C.3)–(C.5), and a single conversion requires  $(b + 1)$  clock periods. This additional buffering will add to the overall jitter of the sampling network but can be minimized with a sufficiently precise initial



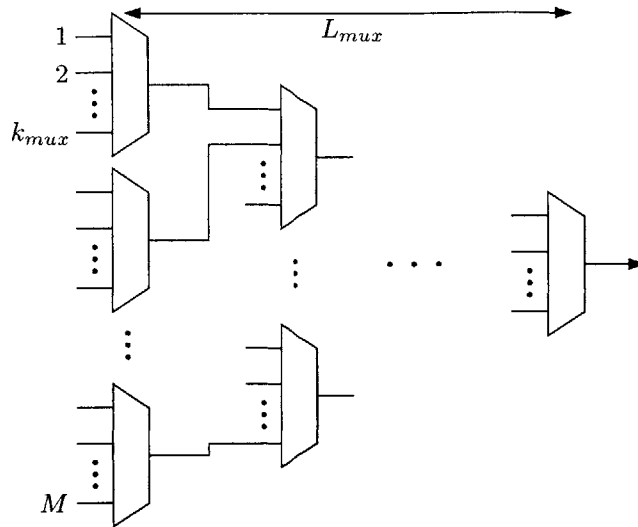


Figure 5-8: Mux tree to combine  $M$  channel outputs into a single ADC output, where each line in the figure is a  $b$ -bit bus. Each stage reduces the number of outputs by  $k_{mux}$ .

clock and using a quiet voltage supply for the clock buffer chain, which was the approach chosen in the final testchip.

It is important to note that (5.7) only applies if the minimum frequency clock is distributed to every channel. If, instead, the sampling clock is distributed in a balanced path to every channel, then (5.6) still applies for the path length, but the number of clock edges per conversion increases from  $b + 1$  to  $M$ , giving a total clock energy that grows as  $M^{3/2}$  instead of  $M^{1/2}$ . Clearly, minimizing the frequency of the clock distribution is critical in interleaved ADCs.

The final clock distribution described in Section 6.2.1 does not scale exactly with this model because the clock paths to different channels vary in length and delay, but it is only applicable to values of  $M$  that are multiples of  $(b + 1)$ , and this simple model with  $C \propto \sqrt{M}$  is still a reasonable, slightly conservative approximation.

### 5.3.7 Output Mux

For a typical application, the outputs of the individual channels must be combined into a single output stream for the ADC. A scalable solution for this sorting is to use a tree of

multiplexers as shown in Fig. 5-8, where each stage in the tree reduces the number of outputs by  $k_{mux}$ . The total number of stages that the output of a single channel passes through is

$$L_{mux} = \log_{k_{mux}} M, \quad (5.8)$$

and the total energy of the output mux is

$$E_{mux} = bL_{mux}C_{sw,mux}V_{DD}^2. \quad (5.9)$$

The prefactor  $b$  accounts for the  $b$  digital bits output by each channel, and  $C_{sw,mux}$  includes the switched capacitance of the mux itself plus the relatively large wiring capacitance in between successive stages.

### 5.3.8 Input Buffer

The most significant analog overhead for increasing the number of interleaved channels is the input capacitance that grows with increased routing and parasitic capacitance of the sampling switches. If the input capacitance  $C_{in}$  of the ADC cannot be driven by the preceding stage in the system, an input buffer is required. For the UWB system considered in this work, the final baseband amplifier has an output impedance of  $R_S = 50 \Omega$ , and therefore an input buffer is necessary only if

$$C_{in} \geq \frac{1}{2\pi R_S f_S / 2}. \quad (5.10)$$

If this condition is met exactly, the input will be attenuated by 3 dB at Nyquist, which is too much attenuation for many applications. In practice, this condition should be exceeded by a fair margin (4 to 10), which can be modeled by adding a factor of  $1/k_{backoff}$  on the right side of (5.10) to designate the target ratio of the input bandwidth to the Nyquist frequency. Appendix C.8 includes the equations when an input buffer is necessary.

The model for the input buffer is simple, neglecting such effects as noise and linearity. Because the final level of parallelism is below the point where an input buffer is necessary, a

more accurate or detailed model is not included.

### 5.3.9 Charge Pump

A charge pump is introduced to allow a decoupling of the sampling voltage from the core supply voltage, as discussed above in Section 5.3.2. In the model, the energy of the charge pump is directly related to the current of the sampling switch drivers and an efficiency parameter,  $\eta_{CP}$ . The efficiency has been set to 70%, which is reasonable based on published charge pumps [117,118]. The charge pump equations are given in Appendix C.9.

## 5.4 Model Results

The energy model has been implemented using Matlab with the global optimization parameters defined in Table 5.1. Figure 5-9 plots the predicted energy per conversion and operating voltages versus the number of parallel channels. The optimum energy point occurs with 60 channels at a 0.65-V core voltage, and the rapid increase in the right side of the curve corresponds to the input capacitance when an input buffer becomes necessary (see Section 5.3.8). The kink in the curve for small values of  $M$  occurs because of the overhead of the charge pump. When  $M < 13$ , the core voltage is sufficiently high that the charge pump, with its non-ideal efficiency  $\eta_{CP}$ , is not necessary.

One important observation from Fig. 5-9(a) is that the energy curve is fairly shallow, with a broad minimum below the optimum energy point. The broadness of this minimum can be understood by considering the breakdown of the energy among the various ADC blocks, as shown in Fig. 5-10 for  $M=36$ . As  $V_{DDD}$  is reduced, the dominant digital portion of the energy decreases, but  $M$  increases significantly (see the flattening of the  $V_{DDD}$  curve at the right of Fig. 5-9(b)). This increases the clock and mux energies, but since they are still a relatively small percentage of the overall power, their significant increase effectively balances the decrease in core digital energy, making the minimum fairly broad.

Instead of trying to operate at the optimum energy point, using 24 or 36 parallel channels

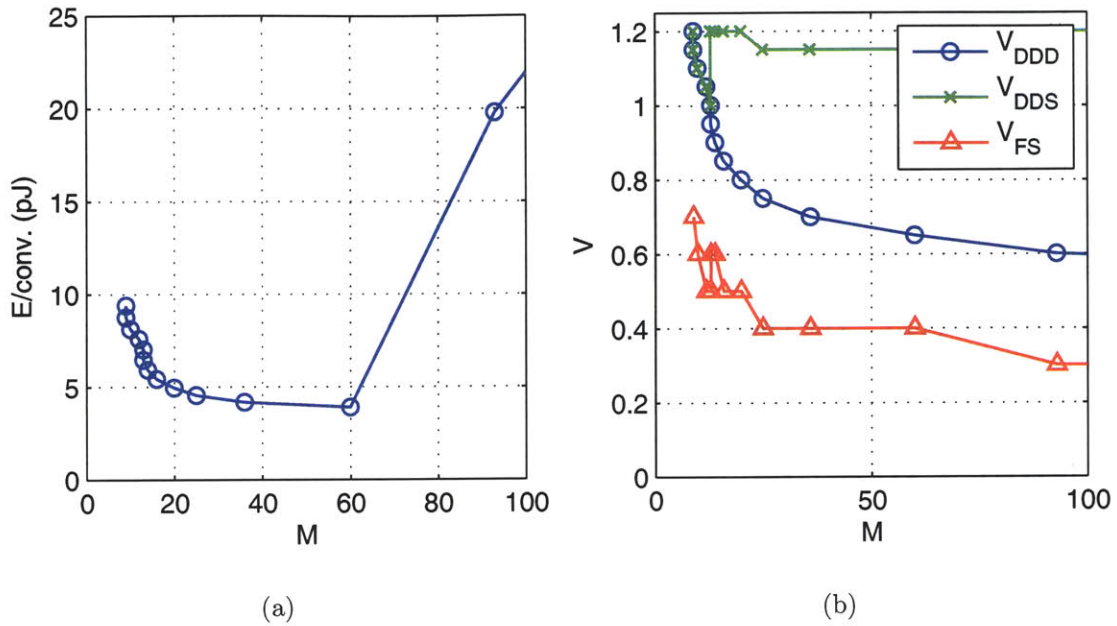


Figure 5-9: Results of the SAR energy optimization. (a) Energy per conversion is plotted versus the number of parallel channels,  $M$ . (b) Optimum voltages for each level of parallelism.

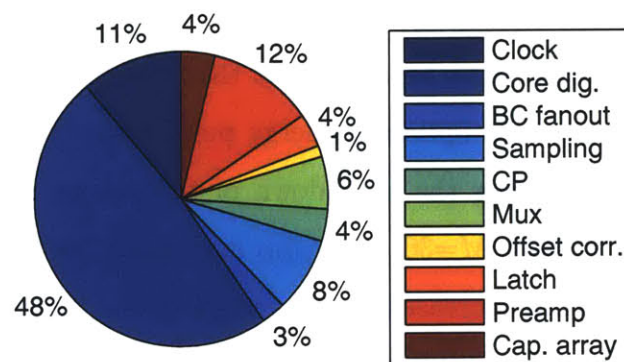


Figure 5-10: Predicted energy breakdown of optimized 36-channel SAR ADC.

Table 5.2: Summary of optimum and 36 channel operating points

Parameter	Global Optimum	36 Channel
$M$	60	36
$V_{DDD} = V_{DDA}$	0.65 V	0.7 V
$V_{FS}$	0.4 V	0.4 V
$V_{DDS}$	1.15 V	1.15 V
$SW_{SAMP}$	N	N
$SW_{REF}$	N	N
$E/\text{conversion}$	3.87 pJ	4.15 pJ

offers significant area and design complexity savings at a cost of 17% or 7% energy penalty, respectively. The final ADC presented in this thesis uses 36 parallel channels to maximize the energy savings while fitting within a 5 mm<sup>2</sup> die. A summary of the optimization results for both the optimum and 36-channel operating points is presented in Table 5.2.

#### 5.4.1 Optimum Energy Point Variations

Several interesting variations can be performed using this model with only a small change in its input parameters. This section presents a few such variations, beginning with the analog input network.

Once an input buffer is required, its energy dominates that of the entire ADC, which limited the optimum number of channels above to that with sufficiently small input capacitance that it could be directly driven with a 50  $\Omega$  source. Not all systems, however, are so constrained in input impedance. Figure 5-11 compares the normal case with those for which an input buffer is always or never used. These cases can easily be included in the model by by setting  $R_S$  in (5.10) to  $\infty$  and 0, respectively. The optimum number of channels is 13 and 121 with and without the input buffer, respectively. Besides increasing the overall energy per conversion, an input buffer significantly adds to the overhead associated with increasing  $M$  because the power of the input buffer grows as  $M^{3/2}$  for large  $M$ , when wiring capacitance dominates over the per channel capacitance. If the entire system were to be optimized, this tradeoff of ADC energy and baseband amplifier output impedance would have a significant

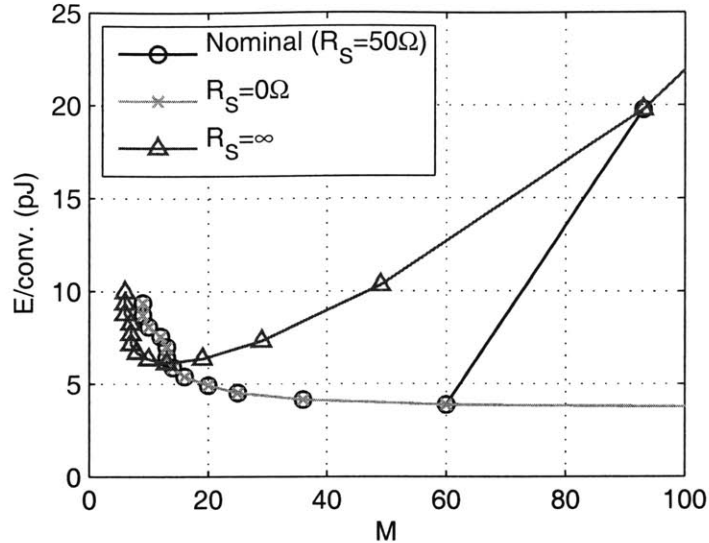


Figure 5-11: Modeled energy versus per conversion versus parallelism with varying source impedances. With an ideal source, an input buffer is never necessary, leading to lower energy and higher optimum value of  $M$  than the case with infinite source impedance, always requiring an input buffer.

impact on the ADC optimum energy point.

Another input signal variant is related to the sampling supply. As mentioned above, for small values of  $M$ , the overhead from the charge pump makes the use of a separate sampling supply voltage inefficient. If a separate sampling supply is not desired or possible, the optimization can be run with  $V_{DDD} = V_{DDS}$ , as compared with both the nominal case and the case with no charge pump overhead in Fig. 5-12. As the supply voltage drops, the sampling switches must rapidly increase in size to be able to track the constant frequency input signal, and at 700 mV this causes a 22% energy penalty for the entire ADC. Below 600 mV, the ADC fails to operate for any value of  $M$  because  $\tau_{sw0} = C_{par0,B}/g_{ds0,B}$ , the inherent switch time constant is too large, making (C.15) negative.

The optimization parameter setup given in Table 5.1 constrains the analog and digital supplies to operate at the same voltage. This is beneficial from a system complexity standpoint because fewer voltage domains are required, but this constraint is not fundamental to the model. An optimization has been performed allowing  $V_{DDA}$  and  $V_{DDD}$  to vary independently. For the 36 channel operating point, the optimum analog voltage is only 50 mV

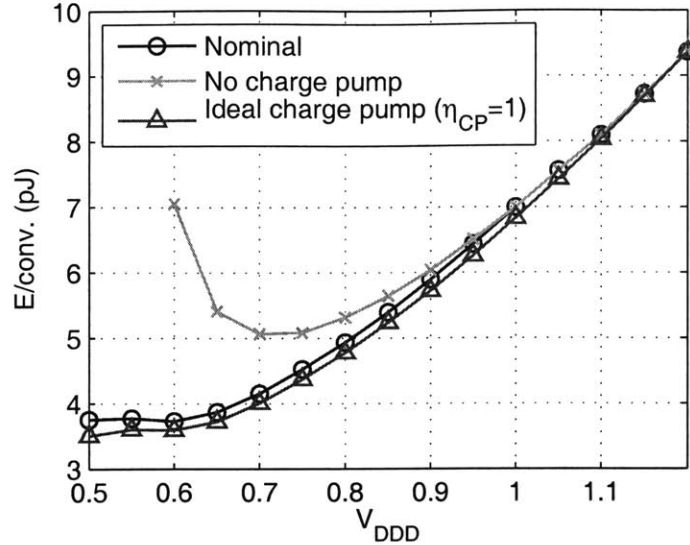


Figure 5-12: Model results comparing the energy versus supply voltage for the nominal, ideal, and no charge pump cases. The charge pump can produce significant overall energy savings as the supply is reduced. Note that the input buffer is not included in this simulation ( $R_S = 0$ ), and the minimum voltage for the case with an ideal charge pump is the minimum voltage allowable in the optimization, 0.5 V.

lower than the optimum digital voltage. The analog voltage supply is limited by biasing requirements in the preamplifier. With a minimum headroom of 150 mV for the tail current source, a supply lower than 650 mV pushes the input device into weak inversion, significantly lowering their  $f_T$ 's. The energy savings for using independent supplies is only 2%, and the consequent increase in system complexity makes this an unattractive choice.

### 5.4.2 Resolution Scaling

The model is designed to be scalable to resolutions other than  $b = 5$  bits, but its applicability is somewhat limited for both very low and high resolutions. At the low end, the model tends to choose device sizes that are impractically small. For example because  $C'_0$  scales as  $b^2$  (2.3), the unit capacitor for 2-bit resolution is only 2.7 fF. At higher resolutions, the effect of circuit noise becomes significant, limiting the model's accuracy to no more than 8 bits, possibly fewer. With that disclaimer, Figure 5-13 presents the energy per conversion step, the optimal number of channels, and the optimal core voltage supply as the resolution is varied. The

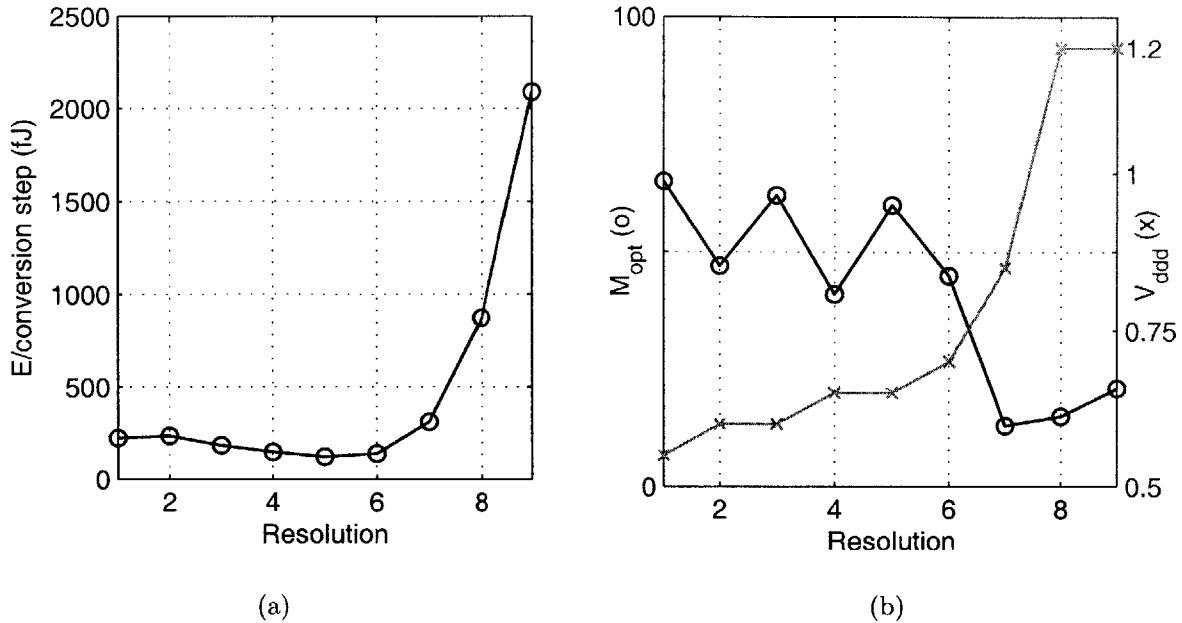


Figure 5-13: Optimum energy point as a function of resolution. The energy per conversion step (a) is the energy per conversion normalized to  $2^b$ . In (b) the optimum amount of interleaving and voltage supply are plotted.

energy per conversion step is the energy per conversion normalized to the ideal quantization step ( $E/\text{conv.}/2^b$ ). This is similar to the figure of merit (1.1) but using  $b$  instead of ENOB. The energy per conversion step is not a monotonic function of resolution; instead, the linear increase in core digital power with the resolution causes an overall decrease in the energy per conversion step at medium resolutions, but at higher resolutions, the matching requirements on the capacitor array and comparator produce an overall increase this metric. As shown in Fig. 5-13(b), as the core digital percentage of power decreases at higher resolutions, the optimal voltage supply increases, along with a corresponding decrease in the amount of interleaving. The jump in  $M$  between 6 and 7 bits corresponds to the resolution when an input buffer is always necessary due to the increase in capacitor sizing. If the input source is ideal, then the optimal supply voltage still increases with the resolution, but the optimum number of channels is far greater (200–300).



### 5.4.3 Architectural Tradeoffs: the VDL

Section 5.3.4 already presented how architectural tradeoffs can be quantified using the energy model presented in this chapter, in that case demonstrating the significant energy savings from adding digital offset correction at the output of every channel. Here, another example will be presented, namely an examination of the energy savings for the VDL-based latch strobe adjustment circuitry presented in Section 4.2.3.

The VDL was inserted to pre-borrow the idle time after the latch has resolved to extend the preamplifier settling time. Considering Fig. 5-7(b), the idle time is the unused time in the second half of the clock period,

$$t_{idle} = t_{per}/2 - t_{d,latch} - t_{setup}. \quad (5.11)$$

Ideally, the VDL uses this idle time to extend the first half of the period, which can be implemented by appropriately modifying the channel block.

At the chosen 36-channel operating point, the VDL increases the length of the first phase of the bit decision period by 21%, but produces an overall energy savings of only 3%. The savings are primarily due to reduced preamplifier current, which is not one of the dominant contributors to power consumption. If the switching energy of the VDL is considered, there are no net savings for using this approach.

### 5.4.4 Conclusion

This chapter has presented a comprehensive mixed signal energy model for use in optimization and evaluation of architectural tradeoffs across the analog/digital boundary. The primary optimization parameters are the amount of interleaving and the supply voltages. Several architectural tradeoffs were analyzed, including digital offset correction, the use of a separate sampling supply voltage, and the VDL-based latch strobe adjustment. The model is set up to facilitate scaling to different technologies, requiring only simulations of the basic building blocks and knowledge of the parasitics and matching. The chosen operating point

of 36 channels is within 10% of the optimal energy point of 93 channels but with significantly less area and design complexity. The next chapter presents the implementation details of this highly-interleaved ADC, with a specific emphasis of the challenges of highly parallel mixed-signal design and the pronounced effects of local variation.

# Chapter 6

## Highly-Parallel ADC With Channel Redundancy

This chapter describes the design, implementation, and measurements of a highly parallel ADC, the final testchip of this thesis. The ADC nominally has 36 channels, which as described in the previous chapter, has energy close to the optimum operating point (60 channels) but with significant area and design complexity savings. In addition, 36 has been chosen because it is a multiple of 12, the value of which will become clear later in this chapter. Even at 36 channels, the design is significantly more complex than the earlier six channel testchips. Primary emphasis is on the mismatch between channels, yield loss due to local variation, and the challenge of synchronization and clocking across the chip. The solutions employed in the testchip for these problems include a hierarchical top plate sampling network with overlapped sampling, redundant channels, and a block structure with strict partitioning of precise and and low power clock domains.

The channels are partitioned into three blocks, as shown in the architectural block diagram of Fig. 6-1. Each block contains twelve nominal and two redundant channels, along with a clock generation unit, a block sampling network, and an output mux. From a black box perspective, each block can be considered its own ADC operating at  $f_S/3$ , and therefore the top-level ADC is actually three time-interleaved blocks, each of which consists of 12 time-

interleaved channels. The channels within a block all share the same  $f_S/6$  clock,  $CLK_{BLK_i}$ , to drive their internal state transitions. As in the previous two testchips, synchronization between channels is achieved by passing along a start token and the blocks synchronize in the same manner. The start token for the first channel is the same as the start token for the block. The signals in bold are at high speeds and require precise timing. These are only distributed to the block sampling network and dividers to generate the lower frequency clocks. The remaining clock distribution is done at the lower core digital voltage. The distance between the block sampling networks is minimized to reduce the wiring capacitance that must be driven at the higher voltage, as described in Section 6.5.

The details of the blocks and channels follow. In Section 6.1, yield loss in parallel circuits and channel redundancy is discussed. The block-level circuits for synchronization between channels (clocking and output mux) are presented in Section 6.2. Section 6.3 examines the error sources from channel mismatch in interleaved converters and presents the sampling network used to counteract timing skew. The channel details are presented in Section 6.4. Finally, Section 6.5 gives the implementation details, and the measurement results are given in Section 6.6.

## 6.1 Redundancy for Yield Enhancement

The energy model in the preceding chapter was designed to give a constant channel and overall ADC performance (distortion, offset, etc.) at every operating point. Many of the constraints, however, such as capacitor matching in (C.11), are statistical in nature and are only meaningful when considered in conjunction with yield for a given specification. This section explores in more detail the yield of parallel systems. For the purposes of this discussion, a conservative definition of yield is used: a chip is considered functioning if the interleaved outputs meet a set of specifications  $\mathcal{S}_G = \{\text{SNDR}_G, \text{ERBW}_G, \text{SFDR}_G, \dots\}$  and every channel meets a set of specifications  $\mathcal{S}_C = \{\text{INL}_C, \text{SNDR}_C, \text{OS}_C, \dots\}$ . These two sets of specifications can be, but do not have to be, identical. The per-channel specifications are necessary to ensure that every sample is accurately converted and avoids some of the

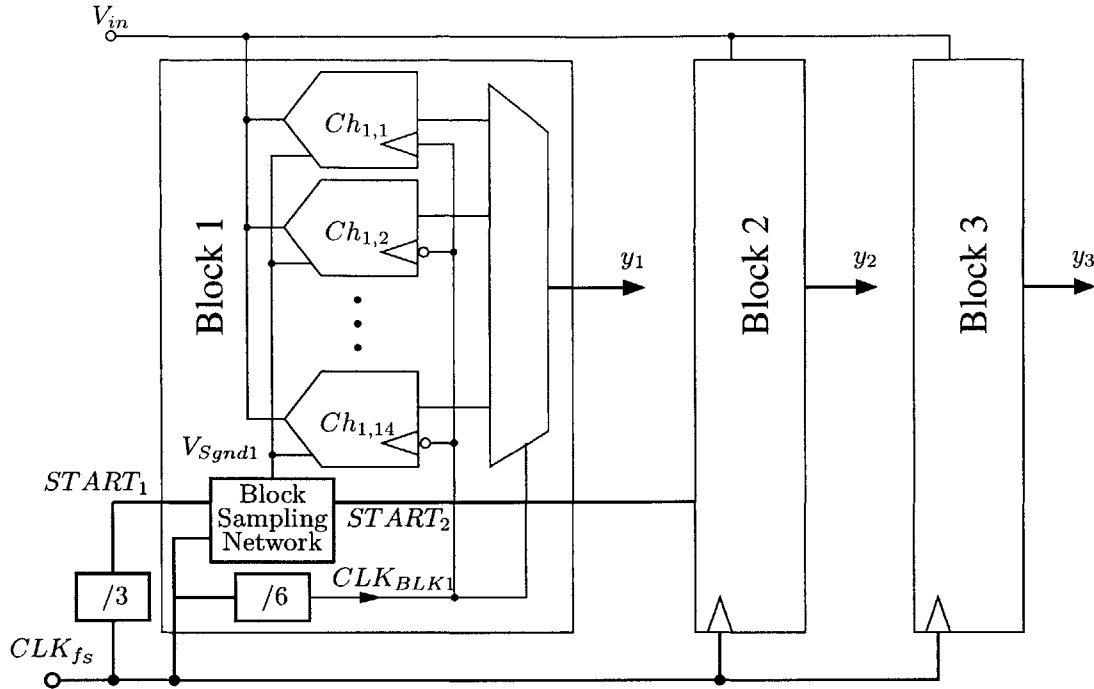


Figure 6-1: Architecture of the final testchip, which consists of 3 blocks of 12 nominal and 2 redundant channels. Each block has its own master sampling network. Only the signals and circuits in bold operate at high speeds; the rest operate at the channel clock rate with reduced timing accuracy.

averaging inherent in the methods used to measure these output parameters. For instance, INL is typically measured using the code density test [85], which relies on a statistical binning of the ADC outputs that is then compared to the expected output code distribution. Any uncorrelated nonlinearities between channels are averaged out, producing up to a  $\sqrt{M}$  improvement in INL even though the accuracy of any particular channel may be much worse.

In general, variations that affect yield are grouped into global or local variations. Global variations, often discretized into process corners, affect all devices across the chip. Local variations represent differences within a die. For parallel systems, global variations are no greater challenge than for non-parallel systems; however, local variations severely impact the yield of parallel systems. Assume that the yield of a single channel ADC can be independently attributed as  $Y_G$  and  $Y_L$  from global and local variations, respectively. Then an  $M$ -way interleaved ADC, would have a yield of  $Y_G Y_L^M$ .

Several ADC parameters are limited by local variation in any technology. For instance, offset and static linearity are usually related to the matching of transistors, resistors, and/or capacitors, where the matching is degraded by local variation between the active or passive devices. In deep sub-micron CMOS, however, random dopant fluctuations have increased local variations for transistors sized relative to the minimum feature size<sup>1</sup> [119], meaning that local variations are a greater concern, influencing not only the traditional parameters but also digital delays, and preamplifier gain/bandwidth products.

Most local variations can be reduced at design time by increased area and/or power. The variance of the threshold voltage in transistors, which impacts comparator offset and digital propagation skews, is inversely proportional to the area [70]. The ratio error in capacitors, which limits the static linearity, is also reduced with increasing capacitor size [67]. The penalty for maintaining yield is significantly increased channel area and the power needed to drive the greater capacitance, which negates much of the benefit from increasing parallelism.

This tradeoff of yield, matching, area, and power can be examined for the case of capacitor array matching, which directly impacts the static linearity of a channel. Figure 6-2 plots a set of waterfall curves of the yield for a given INL specification versus matching of the unit capacitors within the array. To the right, matching degrades and the yield falls off accordingly. In addition, as the number of parallel channels increases, the overall yield drops rapidly for the same capacitor matching. To maintain the same yield level  $Y_0$  as the number of channels increases from 1 to 36, the capacitor matching must improve from  $x_1$  to  $x_{36}$ . This corresponds to a  $2.5\times$  increase in capacitor area. Not only does this lead to a significant increase in an already large die because the capacitors are the plurality of the channel area, but it also leads to an increase in power consumption as larger capacitors require larger switches and higher digital power consumption.

Another technique to improve linearity is to use calibration. The well known SAR calibration procedure [31, 120] introduces a resistive calibration DAC coupled to the output of the main DAC through a capacitor. This technique can fully calibrate any static linearity

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<sup>1</sup>This does not only refer to minimum sized devices; instead, it refers to scaling a potentially larger device between successive technology generations in proportion to the change in minimum feature size.

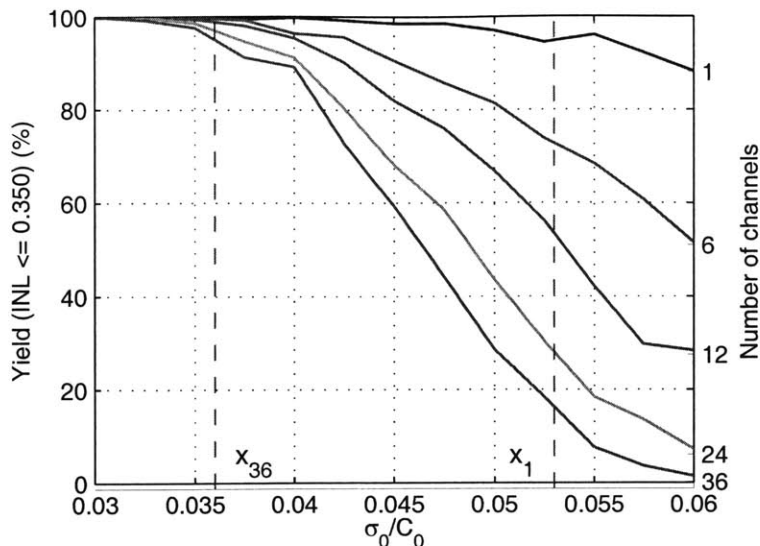


Figure 6-2: Yield versus capacitor array matching as number of channels increases.  $\sigma_0/C_0$  is the relative variation of the unit capacitors within the array. Yield is every channel having an INL better than a 0.35 LSB specification.

and offset errors of the DAC and comparator, but it introduces a complexity and energy overhead to every channel. In addition, some of the local variations that are more significant in deep sub-micron CMOS, including digital propagation delays and timing skew, cannot be fixed with this procedure. Calibration techniques can be designed for each of these error sources, but this leads to rapidly increasing channel complexity with associated implications on the design and test time of the ADC.

Instead, this work proposes the use of redundancy to mitigate yield loss from local variation in this parallel ADC. Memories have long used redundancy to correct failures in large arrays [121] and, more recently, for column sense amplifier offset reduction [122]. Redundancy is also not new to ADCs; redundant comparators have improved linearity in the flash architecture [123, 124], but nothing as extensive as redundant channels. A simple example demonstrates the advantage of redundancy. Assume a comparator occupying area  $A$  has an offset that is normally distributed with standard deviation  $V_{OS0}$ , and the yield constraint

requires the offset to be less than  $V_{OSmax}$  (Fig. 6-3(a)). Then the total yield will be

$$Y_A = 1 - 2\Phi\left(-\frac{V_{OSmax}}{V_{OS0}}\right), \quad (6.1)$$

where  $\Phi(x)$  is the cumulative distribution function (CDF) of normal random variable  $x$ .

If now the available area is increased to  $2A$ , both sizing and redundancy can be used improve the yield. By doubling the transistors' areas, the offset standard deviation decreases by  $\sqrt{2}$  (the narrower curve in Fig. 6-3(a)) and the yield is

$$Y_{sizing} = 1 - 2\Phi\left(-\frac{\sqrt{2}V_{OSmax}}{V_{OS0}}\right). \quad (6.2)$$

If instead there are two comparators of size  $A$ , and the one with the lowest absolute offset is selected, the expected yield is

$$Y_{redundancy} = 1 - 4\Phi^2\left(-\frac{V_{OSmax}}{V_{OS0}}\right). \quad (6.3)$$

These yields are plotted in Fig. 6-3(b) for various  $V_{OSmax}/V_{OS0}$  ratios. As expected, both sizing and redundancy improve yield compared to to the original comparator, but redundancy always produces the higher yield.

Redundancy is most effective when it is applied at the highest level of parallelism of a circuit or system. In this thesis, redundancy is applied at the level of the channel. This does not require complete *a priori* knowledge of the impairments that will reduce yield; instead, any of offset, digital timing delays, static linearity, dynamic linearity, or skew can be measured during initial chip testing, and the best channels chosen. Redundancy is applied to maintain the per channel requirements on par with those of the initial two prototypes.

While the offset requirements can be easily described with closed form solutions, as shown above, other specifications (e.g., timing constraints or static linearity), have more complex distributions that are not representable by simple equations. Instead, Monte Carlo behavioral simulations are used to determine the effect of redundancy on these other parameters.



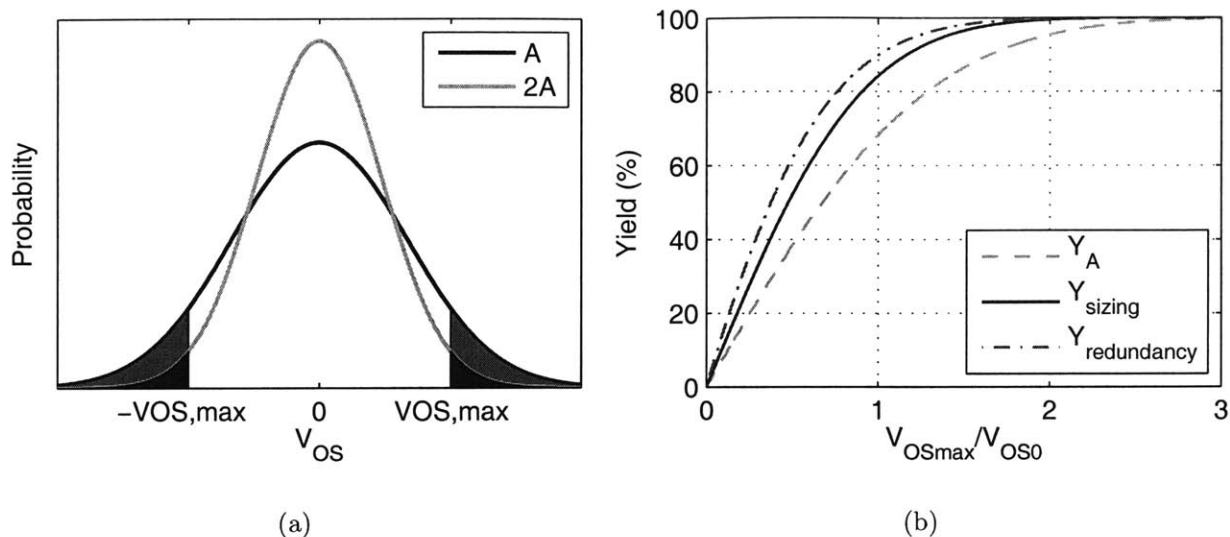


Figure 6-3: Comparison of sizing and redundancy to improve yield when the area of a comparator is doubled. (a) The offset distribution for two different total comparator areas. (b) Comparison of the yield (offset within  $\pm V_{OS,max}$ ) versus normalized offset standard deviation for the sizing- and redundancy-based yield improvement strategies.

As a particular example here, we will show how redundancy affects unit capacitor size requirements to meet a certain INL.<sup>2</sup> Figure 6-4 revisits the capacitor array sizing example but with the addition of redundant channels. With six redundant channels, the yield lies close to the single channel yield curve. For a constant yield, using six redundant channels allows more than a  $2\times$  reduction in the unit capacitor size versus the case for no redundant channels. Redundancy introduces a 17% area overhead for the extra channels, but each channel is smaller. This demonstrates that redundancy can have only a small area cost. In addition, these results can be fed back into the energy model, which did not account for the increased capacitor size as the number of channels increases. If the unit capacitance is doubled in size, the energy model predicts a 15% energy penalty. For the same yield, redundancy results in a lower energy than sizing.

The overall yield of the ADC due to i.i.d. local variations among the channels is a Bernoulli

<sup>2</sup>As shown in Appendix B, the INL for a specific output code is a Gaussian distribution and therefore can be analyzed using the same analysis as above. The overall INL of the ADC, however, is the INL of the worst case output code and has a more complex, non-Gaussian distribution.

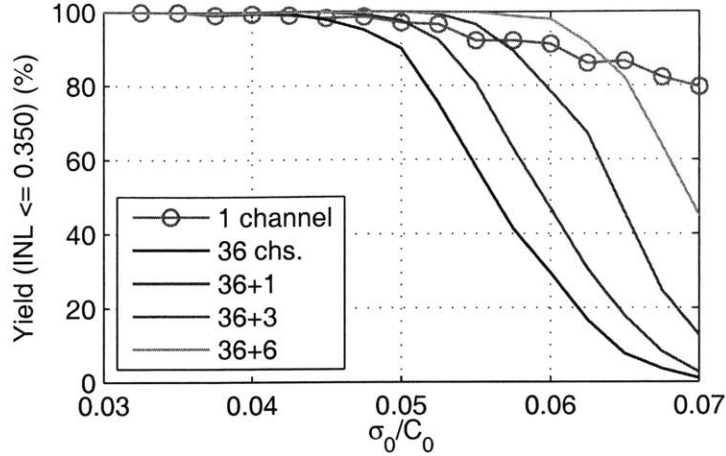


Figure 6-4: Yield versus capacitor array matching as number of redundant channels increases. Also shown is the case for a single SAR ADC.

random process. If the probability of a single channel working is  $Y_L$ , then the probability of at least  $M$  out of  $M_{total} = M + M_{red}$  channels is

$$Y_{red} = \sum_{j=M}^{M+M_{red}} \binom{M_{total}}{j} Y_L^j (1 - Y_L)^{M_{total}-j}. \quad (6.4)$$

Figure 6-5 shows the required channel yield  $Y_L$  as a function of the number of redundant channels with 36 nominal channels. For a target yield of 95%, using six redundant channels lowers the required per-channel yield from the stringent specification of 99.86% to the much more relaxed 91.9%.

The translation of this relaxed specification into the channel design is heavily dependent on the particular metrics that are important. If area is the primary concern, the capacitor arrays, as the plurality of the channel area, are the appropriate choice for downsizing in response to the relaxed yield, as described above. In the design of this chip, the capacitors were made as small as permitted by the DRC and LVS tools but still achieved sufficient matching that they are not the primary limitation on yield. Instead, the relaxed yield was reserved for faults, such as residual timing skew and digital propagation delay errors, that are both not as correctable and not fully modeled during the design phase.

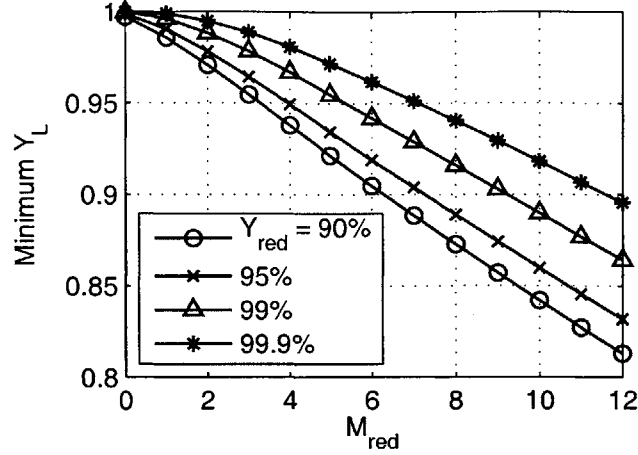


Figure 6-5: The required value of the per-channel yield  $Y_L$  to meet a target overall yield  $Y_{red}$  as a function of the number of redundant channels. The nominal number of channels  $M$  is 36.

## 6.2 Block Details

### 6.2.1 Clock Generation and Block Redundancy

As discussed in Section 5.3.6, it is critical to distribute the minimum frequency clock to every channel. As every channel requires  $(b + 1)$  internal clock periods for a conversion,  $M$  clocks of frequency  $f_{ch} = (b + 1)f_S/M$  separated in phase by

$$\Delta\phi = 2\pi(b + 1)/M \tag{6.5}$$

radians are required. Generating  $M$  phase shifted clocks leads to significant complexity in the clock generation network; however, if  $k \cdot (b + 1)/M = 1$  for some integer  $k < M$ , then channels  $i$  and  $i + k$  share the same clock phases. In particular, if  $M \bmod 6 \equiv 0$ , as in the first two prototypes, then six channels share the same clock phase, and only six clocks must be generated for the 36-way interleaved prototype. In general, for a  $b$ -bit time-interleaved SAR ADC, clock generation requirements are minimized if  $M \bmod (b + 1) \equiv 0$ .

The redundancy analysis in the preceding section made the assumption that any  $M$  channels could be chosen out of the total pool of  $M + M_{red}$  available channels; in this thesis,

this is called global redundancy. This is opposed, however, to each channel receiving a single low frequency clock. One solution is to locate the redundant channels nearest to the clock generation block so that they can receive, with minimal routing distance, either a full speed  $f_S$  clock or all 6 of the phase shifted  $f_S/6$  clocks. Still, this approach does not scale well with larger numbers of redundant channels, and synchronization using the simple start token passing scheme remains challenging. More sophisticated and potentially power hungry synchronization methods would have to be developed for the redundant channels.

Another approach is to use block-level redundancy, where the channels are initially grouped into blocks sharing common clock phases, and then redundant channels are allocated to each block. Assume that each block has  $M_{block}$  nominal channels and there are  $N_{block} = M/M_{block}$  blocks. By allocating redundancy on a per block basis, instead of globally, the yield is degraded for the same total number of redundant channels. The yield for a single block is given by substituting the appropriate number of nominal and redundant channels within a block into (6.4), which produces

$$Y_{single\ block} = \sum_{j=M_{block}}^{M_{block+r}} \binom{M_{block+r}}{j} Y_L^j (1 - Y_L)^{M_{block+r}-j}. \quad (6.6)$$

Here,  $M_{block+r} = M_{block} + M_{red}/N_{block}$  is the total number of channels in a block. The total number of redundant channels is now constrained to be a multiple of the number of blocks. The total yield for block level redundancy is the product of (6.6) for each of the blocks

$$Y_{block,red} = \left[ \sum_{j=M_{block}}^{M_{block+r}} \binom{M_{block+r}}{j} Y_L^j (1 - Y_L)^{M_{block+r}-j} \right]^{N_{block}}, \quad (6.7)$$

Redundancy is most effective when the block size is maximized (i.e., global redundancy). Reconsidering (6.5) with  $M = 36$  and  $b = 5$ , channels  $i$  and  $i + 6$  share the same clock phase as mentioned above, but channels  $i$  and  $i + 3$  differ in clock phase by  $\pi$  radians. The clock for channel  $i + 3$  can therefore be derived from the clock for channel  $i$  using a digital inverter. The  $f_S/6$  clock distributed to the channels should have a 50% duty cycle, which is

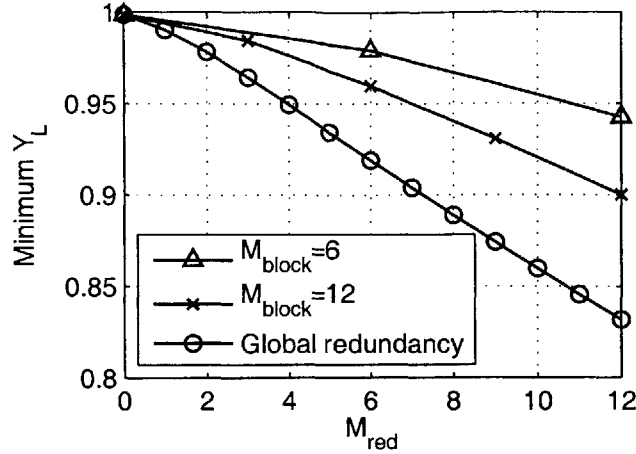


Figure 6-6: Comparison of global and block redundancy. Plotted is the required channel yield for a 36-way interleaved ADC with a target yield of 95%.

straightforward to ensure. This allows setting  $M_{block}$  to 12 without increasing the frequency of the clock that is distributed within the block. Figure 6-6 compares the required channel yields  $Y_L$  for global and block redundancy. With six redundant channels, the required channel yield for block redundancy with  $M_{block} = 12$  is 96%, higher than for global redundancy but still much lower than the yield required without redundancy. Considering the complexity advantage, this chip uses six redundant channels split evenly between three blocks with 12 nominal channels (cf. Fig. 6-1).

The 14 channels within a block are arranged in a chain with a fixed order. The alignment to the proper clock edge and synchronization between channels is performed using the circuit shown in Fig. 6-7. During normal operation, 12 channels have  $EN_i$  asserted. The positive or negative clock edge for the 50% duty cycle  $f_S/6$  clock,  $CLK_{BLK}$ , is selected by  $INVCLK_i$ , which is negated between successive enabled channels. In addition, an enabled channel delays the  $START$  signal by half of an  $f_S/6$  clock cycle. A disabled channel is transparent to both the  $START$  and  $INVCLK$  chains, and  $INVCLK_0 = 1$ . In this fashion, the clock and synchronization overhead for redundancy is minimal.

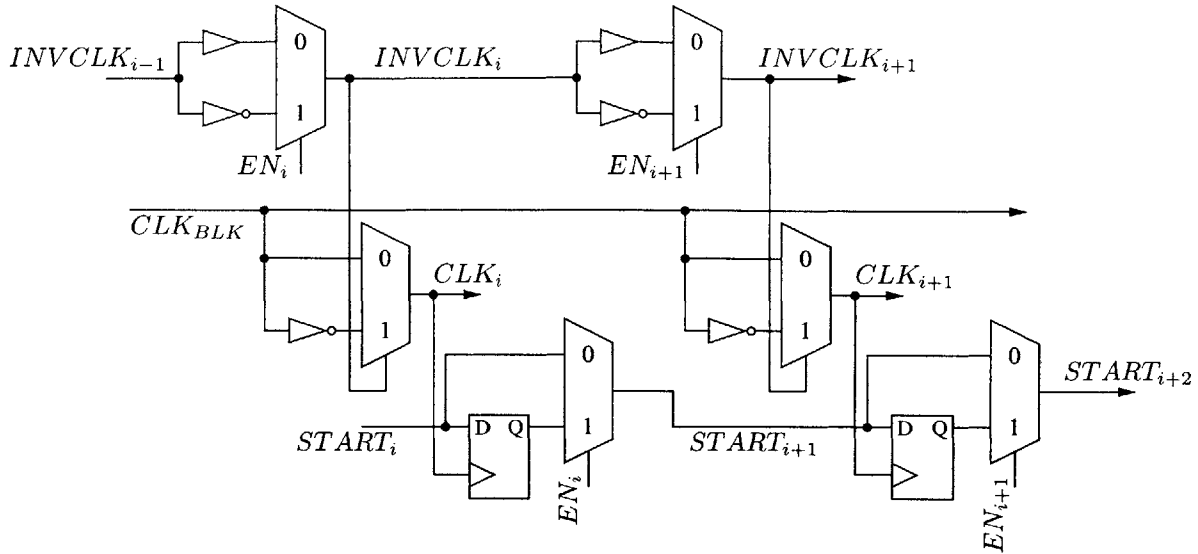


Figure 6-7: Synchronization circuitry of the channels within a block. Successive enabled channels align their clock,  $CLK_i$ , to opposite edges of the  $f_S/6$  clock,  $CLK_{BLK}$ , and delay the  $START$  token by half a clock period.

## 6.2.2 Clock Partitioning

The  $f_S/6$  block clocks are designed to be distributed at the low core digital voltage in order to further save clock distribution power. It is not possible to maintain precise timing alignments with regards to skew and jitter at these lower voltages. Fortunately, the hierarchical top-plate sampling network described below obviates the need for the channel clocks to have precise timing. The precise clock is available at the master sampling switch (see Fig. 6-1), and only one precise clock edge is sent from the master sampling network to a channel to indicate the end of sampling. Sending one precise edge per conversion to each channel is more efficient than distributing the precise clock to every channel; however, every channel needs its own balanced path to the master sampling switch to receive this edge.

## 6.2.3 Output Mux

The output mux is the last major overhead for highly parallel ADCs. To ease PCB design and speed of test equipment, the outputs of the three blocks are kept separate. The output

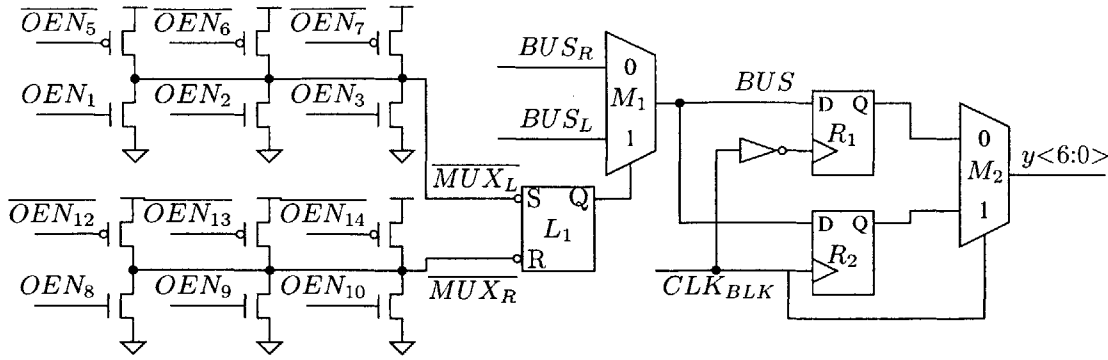


Figure 6-8: Diagram of the block output mux.  $BUS_L$  is shared by channels 1–7, and  $BUS_R$  is shared by channels 8–14. All datapath signals are 7 bits wide. Left/right select signals  $\overline{MUX_L}$  and  $\overline{MUX_R}$  are dynamic nodes charge and discharged by the appropriate channel output enable signals,  $OEN_i$ .

mux combines the outputs of the 12 active channels within a block to a single bus operating at  $f_S/3$ . Within a block, the output mux is implemented as a shared bus. At the end of a conversion, the channel drives its digital output onto the mux bus for one half of the  $f_S/6$   $CLK_{BLK}$  clock period. The mux bus driver is high impedance during the rest of a conversion. The bus is sampled on both the rising and falling edges of  $CLK_{BLK}$ . As each channel outputs only when its data is ready, channels disabled due to redundancy are automatically skipped.

Routing the mux bus to all 14 channels within a block increases the parasitic capacitance that must be driven by each channel. The bus is partitioned into two separate buses, serving channels 1–7 and 8–14, respectively, as shown in the output mux schematic in Fig. 6-8. The datapath within the mux is 7 bits wide to allow for digital offset correction down to 1/4 LSB. The set-reset latch  $L_1$  controls whether the left half of the mux (channels 1–7) or the right half of the mux (8–14) is being sampled by the double edge triggered flip flop, formed from the two single edge triggered flip flops  $R_1 - R_2$  and two-input mux  $M_2$ .  $L_1$  is set when the output enable  $OEN_i$  of channels 1–3 is high, and it is reset when the output enable of channels 8–10 is high. The output enables from channels 3 and 10 are required because it is possible that channels 1–2 or 8–9 are disabled due to redundancy. The signals  $\overline{MUX_L}$  and  $\overline{MUX_R}$  are routed along with the entire bus, which allows  $M_{N1}-M_{N6}$  and  $M_{P1}-M_{P6}$  to be placed near the outputs of the appropriate channels, minimizing the capacitance that must

be driven by  $OEN_i$ . Partitioning of the output mux beyond the two halves is possible to further reduce the bus capacitance but at increased depth of the mux tree.

### 6.3 Error Sources in Interleaved ADCs

Time-interleaved converters are subject to impairments not present in single-channel ADCs. Mismatches in the offset, gain, and sampling instants produce distortion; an analysis of the reduced SNDR and spurious tones can be found in [125]. A number of calibration procedures have been proposed to reduce these mismatches. In [47], each of the channels is calibrated to have equivalent gain and offset to a reference ADC. To enable background operation, therefore not interrupting conversion, an extra high-speed channel is included. The digital background calibration in [50] injects a pseudo-random sequence into the input of the ADC, and then correlates to the same sequence to calibrate the gain and offset of the two time-interleaved ADCs to each other. This technique increases the required input dynamic range of the ADC to handle the dither signal added to the maximum analog input.

By using pseudo-random chopping (swapping the differential paths) of the input, the calibration in [53] does not restrict the input dynamic range. This work further calibrates for timing skew, as well as gain and offset mismatches. The skew compensation circuit uses a 21-tap FIR filter that was implemented off chip.

When considering the low resolution of the ADC for this thesis, any calibration algorithm must be carefully considered for its impact on the overall power consumption. For this purpose, I will differentiate between correction and calibration circuits. The former necessarily run continuously, actually correcting the digital outputs. These would be an adder, a multiplier, and a filter for offset, gain, and timing mismatches, respectively. The calibration circuits determine the proper values for correction and can be run at a much reduced duty cycle, sleeping after the values have been chosen. The minimal duty cycle limits the average power consumption of the calibration loops, whereas the correction circuits directly impact runtime power.



### 6.3.1 Hierarchical Top-Plate Multi-Sampling Network

Considering the above distinction, it is clear that offset is the cheapest impairment to correct for digitally, with gain and timing mismatches each more power hungry. Thus, only digital offset correction will be included as part of this thesis, as described in the previous chapter. The gain mismatch is not a serious issue for this ADC as all channels share an identical reference voltage, and the error due to capacitor mismatch, as confirmed by measurements, is minimal. Timing mismatch, however, remains a pernicious problem. In the first two implemented prototypes, it was demonstrated that by sharing a common sampling clock, distortion from skew between six channels could be kept beneath the total noise power. This is not expected to scale well to 36 channels. The die area is much larger, corresponding to longer clock paths that are difficult to match. Also, as discussed in Section 5.3.6, distributing the sampling clock to every channel in a balanced clock tree is very power hungry, with clock distribution energy growing as  $M^{3/2}$ . A related problem to timing skew is phase variation in the sampling networks between channels. Variation in the resistance and capacitance values produces phase mismatch between channels, which for a single frequency input manifests itself similarly to timing skew.

The combined effect of phase mismatch and timing skew is often modeled as adding a timing offset  $t_{skew,i}$  to every channel's sampling ideal sampling point. Without better knowledge of the timing skew a priori,  $t_{skew,i}$  is chosen as a Gaussian random variable of zero mean and standard deviation  $\sigma_{skew}$ . For the case of an  $M$ -way interleaved ADC the maximum SNDR due to this timing skew is [126]

$$\text{SNDR} = 20 \log \left( \frac{1}{\sigma_{skew} 2\pi f_{in}} \right) - 10 \log \left( \frac{M-1}{M} \right). \quad (6.8)$$

This second term is negligible for large  $M$ . The skew should be less than 16 ps<sub>rms</sub> for a 5-bit ADC sampling a 250 MHz input. This is difficult to achieve, particularly as routing length and variation increase across a large die.

One approach that completely eliminates timing skew and phase mismatch is to precede

the time-interleaved ADC with a single high-speed active sample and hold, sampling at the maximum sampling rate [57, 127]. Each channel only sees a held signal, and timing skew does not produce any voltage error if sufficient settling time is used. The active S/H would consume at least as much power as an input buffer, which as mentioned in Section 5.3.8 would significantly increase the overall ADC power consumption.

Gustavsson and Tan proposed a global passive sampling technique that can reduce the skew related distortion by 10–20 dB [128, 129], extending the technique introduced for double sampling ADCs in [130]. They use a hierarchical top-plate sampling network, as shown in Fig. 6-9. This sampling network is similar to earlier hierarchical sampling networks used in multi-step ADC converters that sampled the input onto multiple capacitors at the same time [131, 132]; the network in [128] makes two significant changes: placing the hierarchical network on the top-plate (signal-independent side) of the capacitor and separating the sampling points of the capacitors for use in an interleaved ADC. Compared to the simple sampling network in Fig. 5-5(a), the bottom plate switch connected to  $V_{in}$  is unchanged, but the top plate switch is split into two series switches, a master switch  $S_G$  shared among all the channels, and a per-channel slave switch  $S_{Ti}$ . The timing diagram of the switches is shown in Fig. 6-10. The sampling period for channel  $i$  occurs when both  $S_G$  and  $S_{Ti}$  are closed, with only one channel sampling at a time. The master switch  $S_G$  opens first, defining the sampling instant common to all channels, eliminating timing skew.

The presence of the parasitic capacitor  $C_{par}$ , however, leads to some residual voltage error due to timing skew between the per-channel slave switches. Defining the capacitor divider ratio

$$a = \frac{C_{par}}{C_{par} + C_S}, \quad (6.9)$$

the timing-skew related SNDR is [128]

$$\text{SNDR} = 20 \log \left( \frac{1}{\sigma_{skew} 2\pi f_{in}} \right) - 10 \log \left( \frac{M-1}{M} \right) - 20 \log(a). \quad (6.10)$$

Compared to (6.8), the SNDR improves by  $20 \log(a)$ .

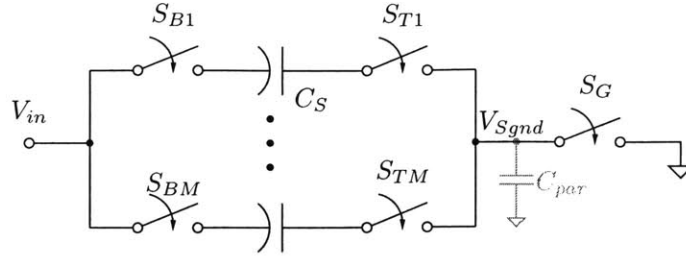


Figure 6-9: Hierarchical top-plate sampling network [128] that cancels skew, with  $C_{par}$  leaving some residual error.

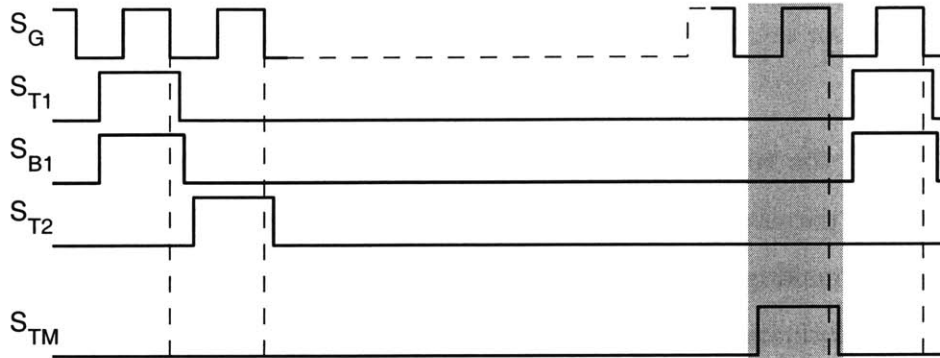


Figure 6-10: Timing diagram for hierarchical sampling network [128]. The shaded region shows the sampling phase for channel  $M$ .

In [24], Gupta, et al., present a hierarchical bottom-plate sampling network (Fig. 6-11) that combines the placement of the master switch on the input node [131,132] with the timing of the switches from [128]. This approach does completely eliminate the skew without any sensitivity to parasitic capacitances, but it suffers two drawbacks. First, when the master switch turns off, it is connected to the input voltage, and it injects a signal-dependent charge onto the sampling capacitor. In [24], even with a bootstrap circuit that linearizes some of this charge injection, the INL was  $\pm 2$  LSB. The second disadvantage involves the tradeoff between switch size, power consumption, and dynamic linearity. The switch connected to the input voltage is larger and more power hungry than the top-plate switch connected to ground. By using two switches in series, the effective width of each switch must double ( $4\times$  total width) to maintain the same  $R_{on}$  and tracking bandwidth. To minimize power consumption, it is desired to increase the smaller top-plate switch by  $4\times$  rather than the floating bottom plate switch.

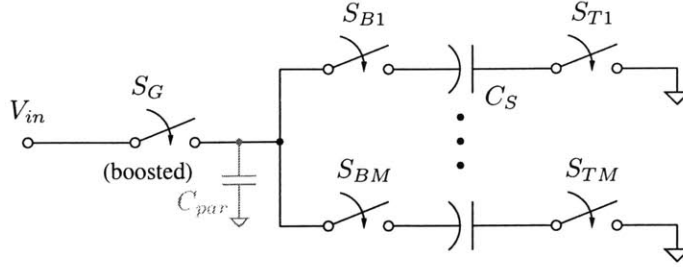


Figure 6-11: Hierarchical bottom-plate sampling network as proposed by [24]. Timing skew between channels is eliminated by having every sampling edge defined by the single master switch. The timing of the switches is the same as that in Fig. 6-10

For this thesis, the residual skew in (6.10) can be made acceptable for a 5-bit ADC by minimizing the parasitic capacitance  $C_{par}$  and the residual skew ( $\sigma_{skew}$ ) between slave sampling switches, making the energy cost for the bottom-plate hierarchical network unnecessary. For the performances achieved in [24], 8.35 ENOB at 500 MHz input frequency, it is expected that the top-plate network is required because  $a$  in (6.9) simply cannot be made small enough. The network of Fig. 6-9 is extended to allow the sampling windows from multiple channels to overlap. Without overlapped sampling windows, the maximum sampling time per channel is less than  $1/f_S$ , but if multiple channels can sample simultaneously, this period can be increased until it approaches the channel clock period  $M/(f_S(b+1))$ . The longer sampling window makes the settling time constraint in (C.12) easier to satisfy; the limiting settling transient is no longer the voltage step on the capacitor array but instead due to clock feedthrough, a second order effect in this differential implementation. If multiple channels are sampling at the same time, it is important to avoid introducing crosstalk through coupling of these sampling networks. Most critically, any noise injected onto the  $V_{in}$  or common switched ground  $V_{Sgnd}$  nodes can corrupt the sampled value on another channel. Figure 6-12 shows the implemented sampling network that minimizes crosstalk during the overlapped sampling periods. A timing diagram of this network is shown in Fig. 6-13, and a comparison of the hierarchical sampling methods with conventional, non-hierarchical sampling is given in Table 6.1.

For each channel, the sampling window is divided into a pre-sampling and a main sam-

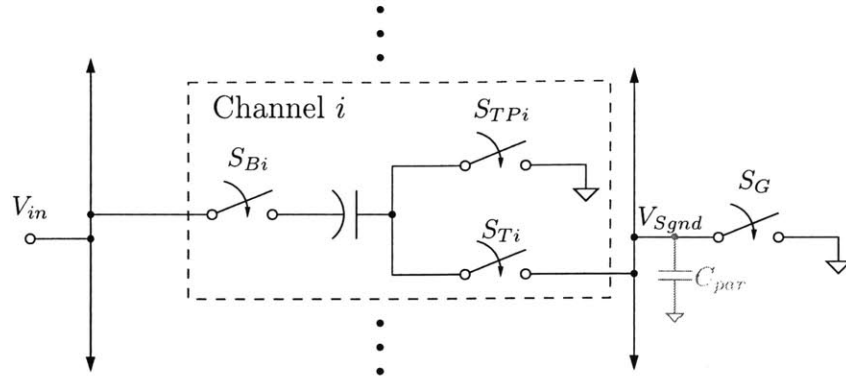


Figure 6-12: Implemented multi-sampling network for use with overlapped sampling windows. The additional top-plate pre-sampling switch,  $S_{TPi}$  avoids crosstalk on  $V_{Sgnd}$ , and  $S_{Bi}$  is turned on with a slow ramp signal.

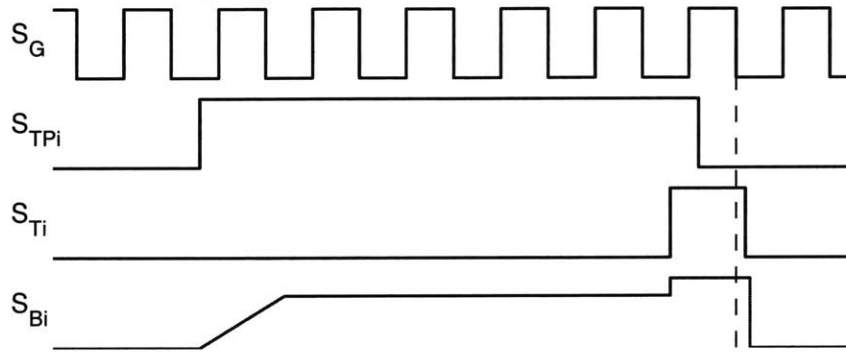


Figure 6-13: Timing diagram of the hierarchical sampling network with slow turn-on bottom plate switch drive and overlapped sampling windows with minimal crosstalk.

Table 6.1: Comparison of interleaved sampling methods

Method	Global sampling switch location	Skew	Switch load	Signal dependent charge injection	Overlapped sampling windows
Conventional	None	Large	Small	No	No
[128]	Top-plate	Small	Medium	No	No
[24]	Bottom-plate	None	Large	Yes	No
Proposed	Top-plate	Small	Medium	No	Yes

pling phase. An additional top plate switch  $S_{TPi}$  is added to every channel that is closed during pre-sampling, connecting the top-plate to ground directly, as opposed to through  $S_G$ . During the main sampling phase for channel  $i$ , switch  $S_{Ti}$  is closed and  $S_{TPi}$  is open, connecting the top-plate to  $V_{Sgnd}$ . Only one channel at a time is connected to  $V_{Sgnd}$ , minimizing

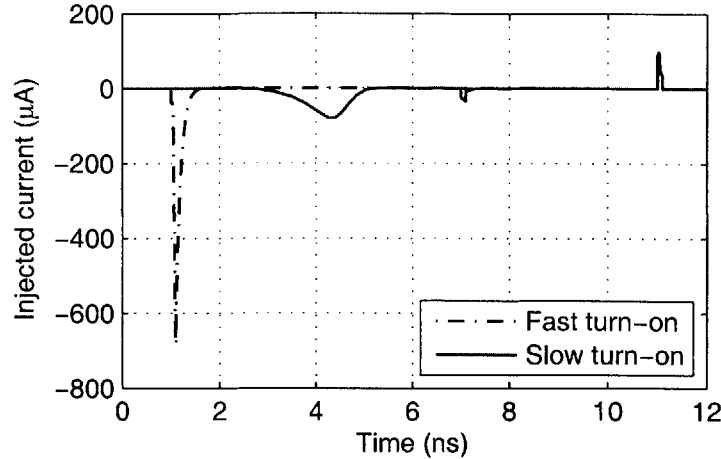


Figure 6-14: Simulation showing the current injected into the input node from the turn-on of the bottom switch when it is driven quickly to  $V_{DD_S}$  and with the implemented slow turn-on gate drive. The slow turn-on reduces the peak current by more than  $8.5\times$ .

crosstalk on this node, and the sampling instant is still defined by the opening of  $S_G$  to reduce skew. The input signal is driven with a  $50\ \Omega$  source, so any current injected onto the input node at the onset of sampling will be converted to a voltage by this resistance, potentially corrupting charge in another channel. The bottom-plate switch  $S_{B_i}$  is turned on with a slow ramp (Fig. 6-13) to the core  $V_{DD_D}$ , and is held there in a weakly on state. During main sampling, the top-plate switch is connected to  $V_{DD_S}$  and is strongly turned on. As shown in the simulations in Fig. 6-14, driving the top-plate switch with this slow turn-on during pre-sampling reduces the current kickback noise onto the input node by  $8.5\times$ , making the resultant voltage error less than an LSB. The kickback noise is sufficiently low that no measures are taken to further align the start of sampling away from critical edges of the other sampling channels. Simulations show that the slow ramp on the bottom plate switch could continue to  $V_{DD_S}$  (without stopping at  $V_{DD_D}$ ) without an increase in kickback noise for a possible simplified implementation.

Figure 6-15 shows the circuit details of the channel sampling logic, used to drive the switches in Fig. 6-12. The rising edge of  $L_{SAMP}$  triggers pre-sampling and strongly turns on the pre-sampling top-plate switch via  $S_{TP}$ . The bottom plate switch is driven using the slow turn-on switch drive composed of  $M_{P_1}-M_{P_3}$  and  $M_{N_1}-M_{N_2}$ . By making  $M_{P_1}-M_{P_2}$  relatively

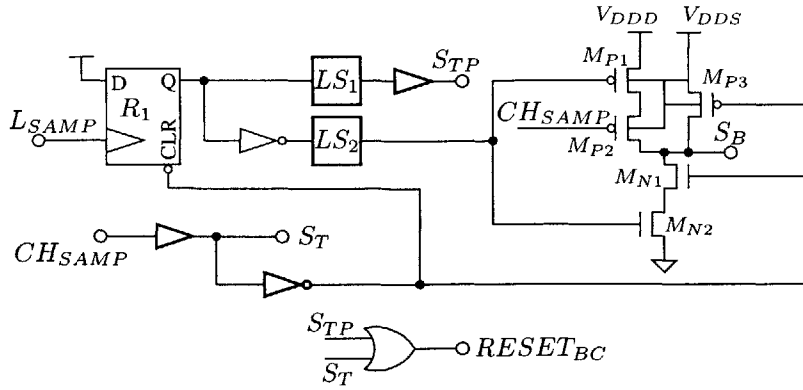


Figure 6-15: The channel sampling logic to drive the bottom plate switch  $S_B$ , and the pre-sampling and main sampling switches  $S_{TP}$  and  $S_T$ . All blocks in bold are powered off of the sampling voltage  $V_{DD3}$ , while the remaining blocks are connected to  $V_{DD3}$ . Level shifters  $LS_1$ - $LS_2$  are used to convert from the lower to higher voltage domains.

weak, output voltage  $S_B$  is slowly driven to  $V_{DD3}$  at the start of pre-sampling. When the channel receives  $CH_{SAMP}$  from the block sampling network (see Fig. 6-16), it turns on the main top plate switch  $S_T$ , connects  $S_B$  to  $V_{DD3}$  via  $MP_3$ , and turns off the pre-sampling top-plate switch by resetting  $R_1$ . Finally, on the falling edge of  $CH_{SAMP}$ , all sampling switches are opened. Level shifters are inserted to convert between the core digital voltage  $V_{DD3}$  and the sampling switch voltage  $V_{DD3}$ .

The signals that produce the falling edge of  $S_{Ti}$  and  $S_{Bi}$  are derived from the falling edge of the master switch signal  $S_G$  using logic connected to the higher voltage sampling supply  $V_{DD3}$ . As digital delay variations increase at lower voltages, connecting this signal path to the higher voltage reduces the skew between the slave switches,  $\sigma_{skew}$ , and improves the SNDR (6.10) at only a small power overhead.

To reduce the distance between the master sampling switch and the channels and to minimize the parasitic capacitance  $C_{Par}$  in Fig. 6-12, three master sampling switches are used, one for each block. This introduces an extra source of skew between the blocks; however, the parasitic capacitance is now less than 1/3 of its value with a single master switch, reducing  $a$  in (6.9), and improving the cancellation of intra-block skew. The precise full speed sampling clock is distributed to all three of these master switches in a balanced layout.

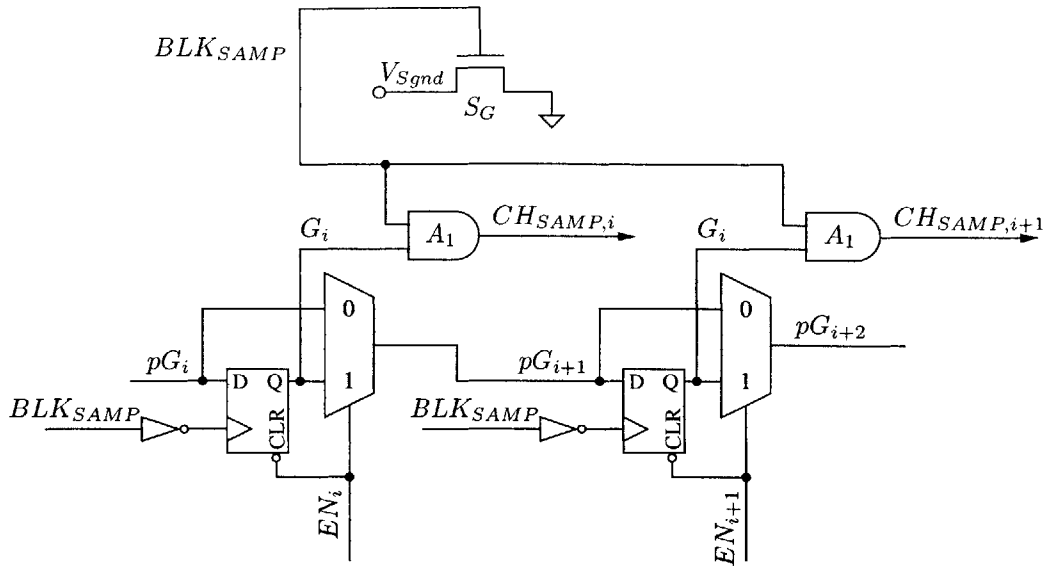


Figure 6-16: Simplified schematic of the per-block master sampling network.  $S_G$  is the master top-plate sampling switch. The sampling clock  $BLK_{SAMP}$  is gated by  $G_i$  to generate the per-channel main sampling window  $CH_{SAMP,i}$ .

Within each block, the master sampling network includes the master sampling switch and the logic to distribute the end of sampling signal to each channel. It operates at the higher sampling voltage  $V_{DDS}$  to minimize skew, jitter, and maximize the overdrive of the sampling switches. A simplified schematic is shown in Fig. 6-16.  $BLK_{SAMP}$  is the  $f_s/3$  master sampling clock derived from the  $f_s$  full speed sampling clock distributed to all three block sampling networks, and it directly drives the global sampling switch  $S_G$ . The AND gates  $A_1$  mask this signal to generate  $CH_{SAMP,i}$ , which is sent only to the one channel that is in its main sampling phase. The falling edge of  $CH_{SAMP,i}$  opens the per-channel top plate switch (Fig. 6-15). The gating signal  $G_i$  is generated with a shift register, similar to that used for the channel synchronization (cf. Fig. 6-7), that is transparent for disabled (due to redundancy) channels. The timing of  $G_i$  is controlled to ensure that it transitions low while  $BLK_{SAMP}$  is low to avoid any glitches on  $CH_{SAMP,i}$ .



## 6.4 Channel Circuit Details

The basic channel structure is similar to the previous two chips but with no analog/digital timing adjustment, either in the form of self-timing or with a VDL.<sup>3</sup> The split capacitor array is used, with the two differential capacitor arrays having swapped subarrays (see Appendix A). The sizes of array switches, buffer chains driving the switches, and transistors in the comparator were initially taken from the outputs of the energy model, with only minor adjustments made after transistor-level simulations. The following sections discuss differences in the channel circuitry compared to the prior chips. These include a half-rate SAR controller, a DAC-biased preamplifier, and the circuits to implement the channel-specific parts of the output mux.

### 6.4.1 DAC-biased Preamplifiers

The duty-cycled preamplifier of Fig. 3-8(b) is enhanced by replacing the fixed current source with a DAC that can be varied on a per-channel basis. This allows post fabrication correction for mismatch in the bias network or PFET load transistors that bias an individual preamplifier out of its desired operating region, which in turn leads to severe offset or linearity degradation. In addition, for channels that have a very low latch offset voltage, the required gain in the preamplifiers is minimal. The bandwidth of the preamplifier is set by the resistance of the PFET loads and load capacitance, both of which are relatively independent of the bias current. By decreasing the preamplifier bias currents in channels with small offsets, the power savings are achieved without sacrificing preamplifier speed or static linearity. For highly parallel systems, the average power can be reduced by allocating power on a per-channel basis; this is an additive effect with redundancy. The price is additional per-channel complexity and a longer test time. If implemented, the potential built-in self test (BIST) described in Section 6.6.4 could also dynamically bias the preamplifiers in response to measured chip performance. For the majority of testing, every preamplifier was

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<sup>3</sup>Section 5.4.3 presented the energy model analysis of the VDL, showing that it does not produce energy savings for this highly parallel converter

biased to the same default value, and only after channel selection is performed were the bias currents adjusted independently (Fig. 6-28).

## 6.4.2 Digital Logic

The SAR logic for the bit decisions is static CMOS as in the second testchip, but it has been redesigned as a half rate architecture, as shown in Fig. 6-17. The  $f_S/6$   $CLK$  (see Fig. 6-7) is initially divided by two to generate  $CLK_{BC}$ . This reduces the switching activity and clock power of the shift register  $R_1 - R_6$ . The rising edge of  $L_i$  denotes the end of the bit decision period for bit  $i = 4, \dots, 1$ . The last bit decision ends at the start of the next conversion (i.e., rising edge of  $L_{SAMP}$ ). Unlike the implementation in Fig. 3-5, the signals  $L_5 - L_1$  do not return low after one clock period. They remain asserted for the rest of the conversion until  $RESET_{BC}$  (see Fig. 6-15) is asserted during the next sampling phase. At the end of bit decision, the switch drive registers  $S_{D4} - S_{D1}$  latch the output of the comparator  $COMP$ , storing it as  $C_i$ . The bottom plate switches of the split capacitor array are driven as

$$S_{5,i}^+ = \overline{C_i} + C_i \overline{L_i} \overline{RESET_{BC}}, \quad (6.11)$$

$$S_{5,i}^- = C_i L_i (\overline{S_{5,i}^+} + S_{5,i}^+ COMP), \quad (6.12)$$

$$S_i^+ = \overline{C_i} L_i, \quad \text{and} \quad (6.13)$$

$$S_i^- = C_i \overline{RESET_{BC}}. \quad (6.14)$$

Before bit  $i$  is determined by the comparator,  $C_i$  is preset high.

Sampling begins on the first rising edge of  $CLK$  after  $START$  is asserted. The divide by 2 that generates  $CLK_{BC}$  (Fig. 6-17) synchronizes to the  $START$  signal, guaranteeing that a rising edge of  $CLK_{BC}$  occurs at this point. After 6 periods of  $CLK$ , or 3 periods of  $CLK_{BC}$ , the result of a conversion is available. The digital output,  $y_{OS}<4:0>$  of the previous conversion is latched by the rising edge of  $L_5$ . The output is then passed to a saturating adder where it is corrected with the predetermined offset value. The output of the adder  $y<6:0>$  is latched by the following rising edge of  $L_5$  and driven onto the shared

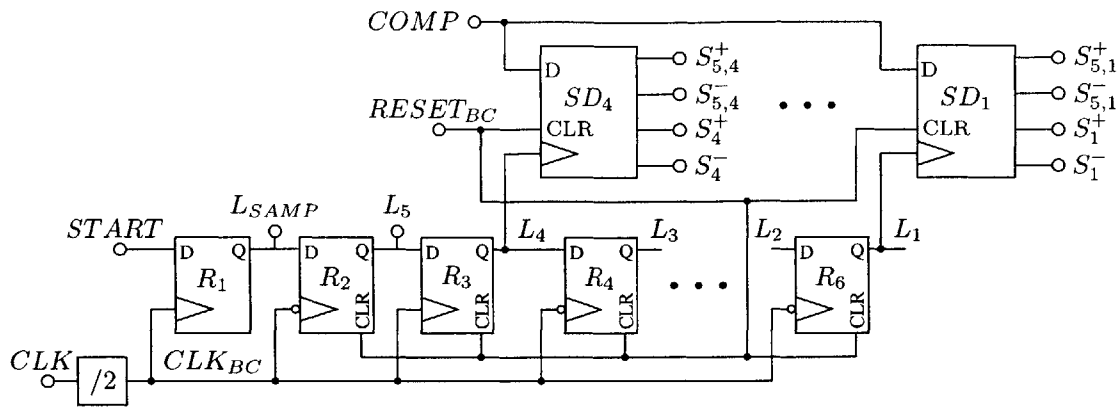


Figure 6-17: Half rate clock architecture for the SAR logic. The rising edge of  $L_i$  ends the decision period for bit  $i$ , latching the output of the comparator  $COMP$  within the switch drive register  $SD_i$ . Half rate is appropriate because no digital computation occurs on the falling edge of  $CLK$ . The outputs  $S_i^\pm$  and  $S_{5,i}^\pm$  directly drive the split capacitor array's bottom plate switches (see Fig 4-7).

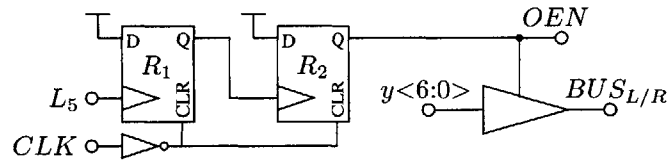


Figure 6-18: Channel output mux driver. The output is enabled for half of a channel clock period, after the rising edge of  $L_5$ .

output bus by the circuit shown in Fig. 6-18. The output enable signal  $OEN$  is delayed by  $R_1$ – $R_2$  after the rising edge of  $L_5$ , which itself is delayed from the rising edge of  $CLK$ . This delay prevents bus contention between two neighboring channels. The falling edge of  $OEN$ , is triggered one half clock cycle later by the falling edge of  $CLK$  through  $R_2$ .

## 6.5 Testchip Implementation

Several configuration options and design for test features have been incorporated on this chip. The primary outputs of each block are synchronized to a common  $f_s/3$  output clock and then driven off chip, along with the output clock. The digital output drivers, shown in Fig. 6-19, can be configured as a low swing differential output or as a full swing single-ended output. In single-ended mode ( $\overline{DIFF}$ ), the negative differential output,  $\overline{PAD}$ , is grounded.

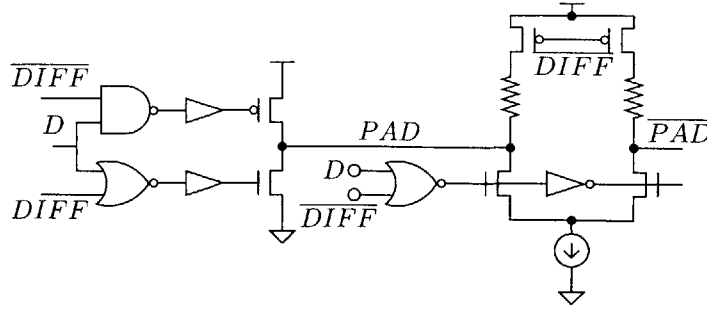


Figure 6-19: Schematic of the reconfigurable output pad. If  $DIFF$  is asserted, the low swing differential output driver is enabled, driving  $D$  to the output, and the single-ended driver has a high impedance output. In single-ended mode, the poly resistor loads are disabled and the single-ended output driver buffers  $D$  to the pad.

This allows direct measurements of ADC performance due to the coupling of the large single-ended output drivers and also facilitates testing with logic analyzers that do not support differential signals. Each block has a separate debug bus that can be used to look at the outputs of a single channel, and this debug bus is directly buffered off chip for the first and the third blocks; insufficient pins were available to do the same for the second block. The debug bus outputs are single-ended because the maximum speed is the channel output clock speed ( $f_S/36$ ). The entire chip is configurable with a 714-bit shift register, which includes the per-channel offsets for digital offset correction, DAC inputs for the per-channel preamplifier biasing circuitry, and channel enables for redundancy.

The basic layout of a block is shown in Fig. 6-20. The channels are arranged in two columns of seven channels each, with the master sampling network located centrally. By using a tall, narrow block, the distance between the blocks' master sampling networks is minimized, lowering the power and skew of the full speed clock distribution. All analog and power signals enter from the bottom of the block, and the output mux and digital outputs are at the top of the block. The routing of all the clock and sampling network signals is located inside the the two columns, and the output mux and debug buses are located on the outside. The analog input  $V_{in}$  and the switched ground  $V_{Sgnd}$  in the sampling network are routed using a balanced tree, and the  $CH_{SAMP,i}$  signals have identical length paths to every channel to eliminate any deterministic skew. In contrast, the  $f_S/6$  clock  $CLK_{BLK}$  is routed

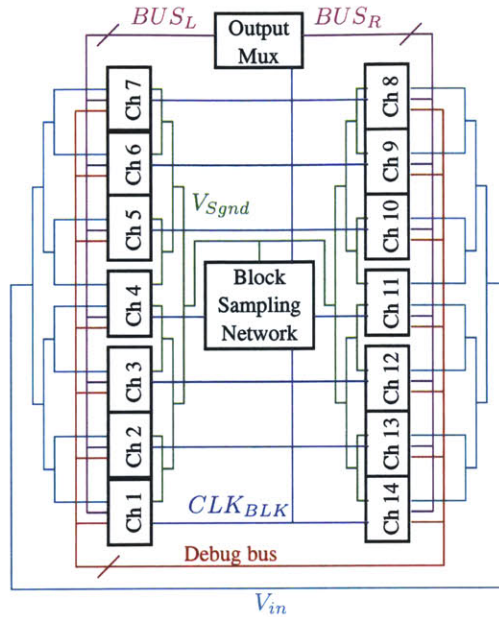


Figure 6-20: Layout of a single block. Not to scale. The gaps surrounding channels 4 and 11 are used to route the clock and input signals between the blocks. Inside the block, the switched ground node for the sampling network is distributed in a tree, and the block clock is distributed with a minimum total path length.

as a signal wire from the bottom to the top of the block, with different effective path lengths to every channel.

The die photograph is shown in Fig. 6-21. The 100-pin die is  $2.65 \times 1.89 \text{ mm}^2$ , and the total active area is roughly  $4 \text{ mm}^2$ . The chip includes several nF of decoupling capacitors for the different supplies.

## 6.6 Measurements

### 6.6.1 Basic Measurements

This section describes a summary of the measured performance and power consumption of the ADC. A detailed discussion of variation, redundancy, and channel selection follows in subsequent sections.

The coefficients for digital offset correction are calculated by inputting a sufficiently busy

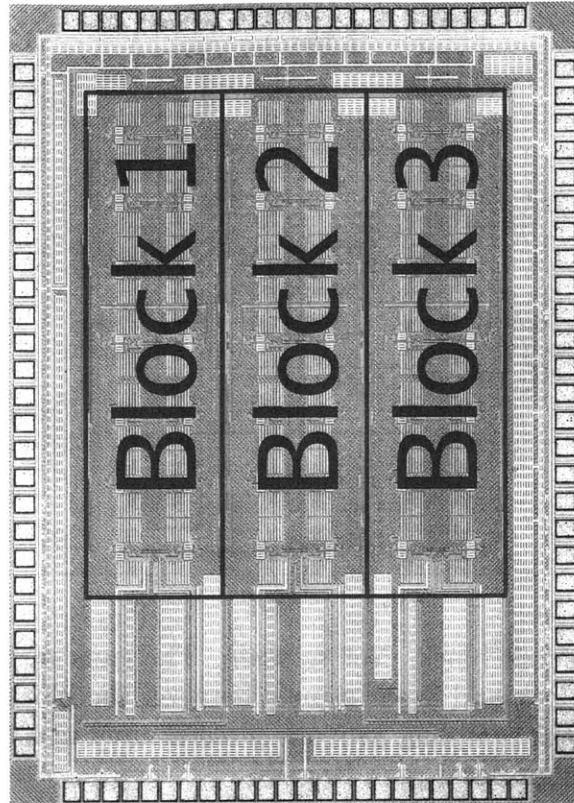


Figure 6-21: Die photograph of final testchip. Analog and power signals enter from the bottom, and the digital signals are output through the top half of the pins.

signal to the ADC and averaging the sorted channel outputs in Matlab. The input signal must be chosen to have no frequency content near any multiples of the channel sampling frequency,  $f_s/36$ . The coefficients are then programmed into the chip, and the accuracy of offset correction can be controlled by Matlab to within 0.25 LSB. Figure 6-22 shows the improvement in SNDR versus level of digital offset correction. The total power in the offset spur train dominates the error power in the ADC if no offset correction is used. Correction past 0.5 LSB provides only minimal improvement in performance. The cost for each additional bit of offset correction is a  $20 \mu\text{W}$  increase in power consumption of the output mux.

As measured using the debug bus, the individual channels are operational up to 9.7 MS/s and 16.7 MS/s at 700 mV and 800 mV, respectively. These speeds correspond to overall sampling rates of 350 and 600 MS/s, as shown in Fig. 6-23. The SNDR drops by roughly

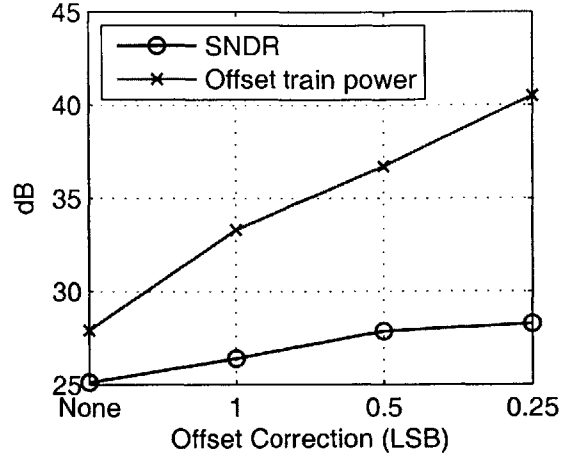


Figure 6-22: ADC performance as the level of offset correction is varied.

4 dB for a Nyquist input (relative to the interleaved sampling rate) as the overall sampling frequency increases from 100 to 600 MHz.

Interleaved measurements are limited to below 400 MS/s operation because of a failure in the output mux. The capacitance on the shared bus ( $BUS_{L/R}$  in Fig. 6-8) was higher than the mux bus driver (Fig. 6-18) was designed to drive. This failure can be partially corrected at test time by adjusting various power supplies to indirectly move clock edges, but this only extends the operating frequency up to 400 MHz. A schematic showing the critical paths and relevant supplies is shown in Fig. 6-24. A separate supply,  $V_{DDC}$ , was pinned out for the clock distribution, which allows somewhat independent control of the clock edges. The channel output registers  $R_1$  and  $R_2$  are clocked by  $L_5$  (Fig. 6-17) and  $CLK_i$  (Fig. 6-7), both of which are driven by  $V_{DD}$  powered blocks. The block clock,  $CLK_{BLK}$ , is powered by  $V_{DDC}$ . Register  $R_5$  is used to synchronize the outputs of all three blocks to a common  $f_S/3$  output clock,  $OCLK$ , in order to simplify testing with a logic analyzer. Both  $R_5$  and  $OCLK$  are powered by the I/O power supply  $V_{DDIO}$ .

In simulations, the bus driver is capable of driving a 200 fF load in 1.6 ns, well below the approximately 4 ns (at 500 MS/s) available between the rising edges of  $L_5$  and  $CLK_{BLK}$ . Parasitic extraction predicted a wiring capacitance of 130 fF, making the design of this block relatively conservative, but measurements indicate setup time failures of  $R_3/R_4$  as the

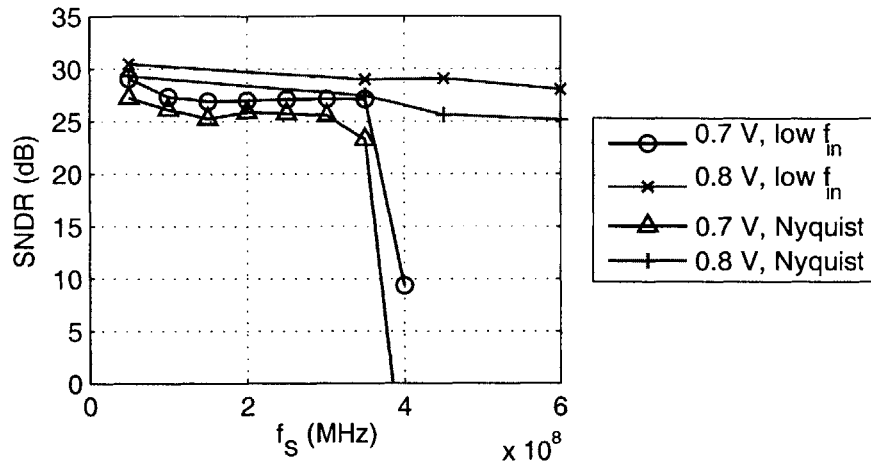


Figure 6-23: Individual channel performance for low frequency and Nyquist inputs. The plotted sampling rate refers to the interleaved sampling rate; the channel is sampling at  $f_s/36$ .

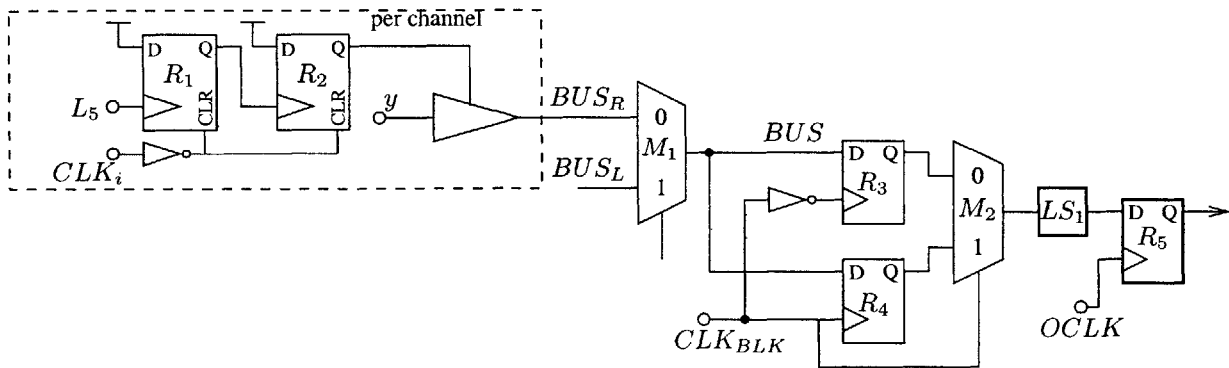


Figure 6-24: Schematic of the path from the output of a single channel to the pad drivers (output of  $R_5$ ). Signals that are not relevant to the critical paths are omitted. The two potential errors are a violation of the setup time of  $R_{3/4}$ , or a violation of the setup time of  $R_5$ . Clocks  $CLK_i$ ,  $CLK_{BLK}$ , and  $OCLK$  are powered by  $V_{DD}$ ,  $V_{DDC}$ , and  $V_{DDIO}$ , respectively.



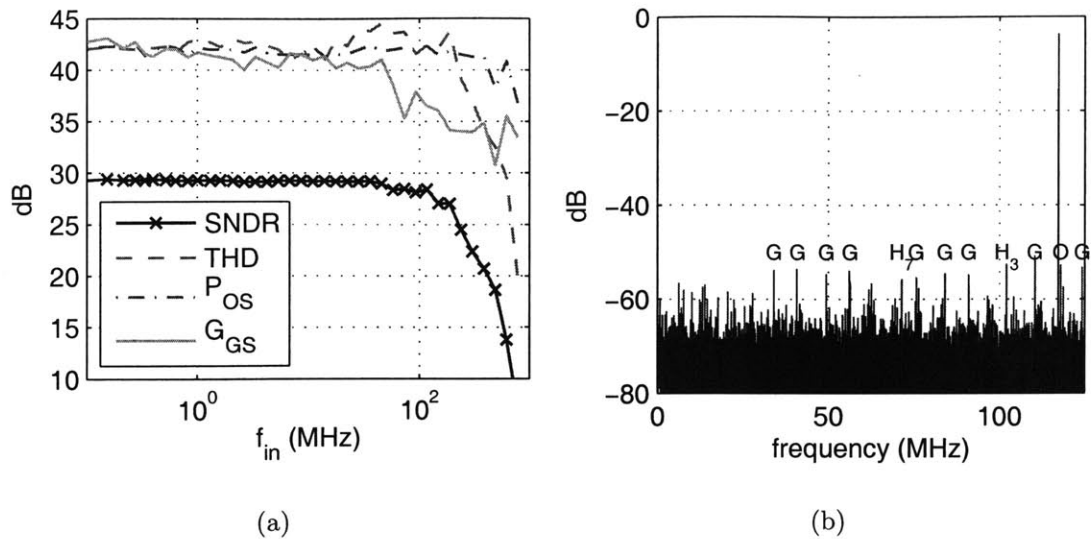


Figure 6-25: Dynamic performance of the ADC at 250 MS/s, showing (a) the SNDR versus input frequency and (b) an FFT of a near Nyquist input. The labeled dominant spurs are harmonic distortion ( $H_i$ ), multiples of the channel sampling frequency ( $O$ ), or mixing products of the channel sampling frequency and the input ( $G$ ).

sampling frequency increases beyond 300 MHz for  $V_{DDD} = 0.8V$ . By lowering  $V_{DDC}$  relative to  $V_{DDD}$ , the clock of  $R_3$  and  $R_4$  is delayed, but this leads to setup time violations of  $R_5$ . The I/O power supply can be lowered to delay the edge of  $OCLK$ , but only a limited reduction in  $V_{DDIO}$  is possible before the output drivers operating upwards of 150 MHz begin to fail. Careful control of  $V_{DDD}$ ,  $V_{DDC}$ , and  $V_{DDIO}$ , has extended the operating frequency up to 400 MHz, but beyond that, output mux failures cannot be completely eliminated and the ADC performance degrades severely.

At 800 mV core voltage, 1.2 V sampling voltage, and 250 MHz sampling frequency, the dynamic performance is plotted versus input frequency in Fig. 6-25(a). The individual error contributions from harmonic distortion ( $THD$ ), total power in the offset train ( $P_{OS}$ ), and the total power in the mixing products from gain mismatch and skew ( $P_{GS}$ ) are plotted separately. The frequency independent part of  $P_{GS}$  is from gain mismatch, whereas its degradation at high frequencies is due to timing skew. An FFT of a Nyquist input tone is shown in Fig. 6-25(b). The ADC achieves an ENOB of 4.42 at Nyquist while consuming total

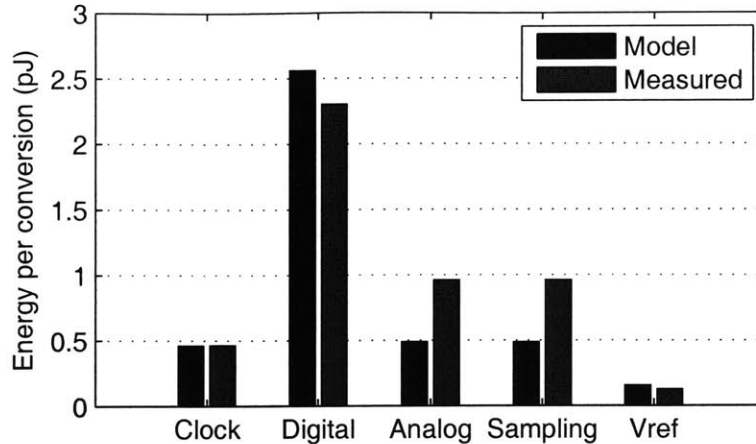


Figure 6-26: Comparison of the measured ADC energy at 250 MS/s with the results from the energy model. Due to the limited number of separate supply voltages, these results do not have the granularity of Fig. 5-10, and the model’s energies have been combined appropriately.

power of 1.20 mW, with a power breakdown as shown in Table 6.2. The energy efficiency of the ADC (1.1) is 240 fJ/conversion step.

Figure 6-26 compares the measured ADC energy at this operating point with the predictions of the energy model. The total energy per conversion is about 13% higher, and the clock, digital, and reference voltages power match well. The measured analog energy is roughly 2× higher than predicted because the ADC is operating at half of the sampling frequency than the energy model was optimized for. While, for a fixed supply voltage, digital energy is independent of clock frequency, analog energy varies inversely with the frequency (see (C.28)). The sampling energy is also higher because  $V_{DD_S}$  has been increased from 1.05 V to 1.2 V, and the first stage of clock distribution to the block sampling networks is driven with the  $V_{DD_S}$  supply, which was not included in the energy model.

At 400 MS/s, the output mux failure requires a core voltage of 1 V with a total power consumption of 2.66 mW (see Table 6.2). The dynamic performance is plotted in Fig. 6-27; the ADC achieves an ENOB of 4.32 at Nyquist, corresponding to 350 fJ/conversion step. The SFDR is degraded at this higher sampling frequency because of increased spurs from the gain and skew mismatches. Functionality is possible at 900 mV with a boosted  $V_{DD_C}$  of 1.1 V (total power consumption of 2.28 mW), but intermittent output mux failure degrades

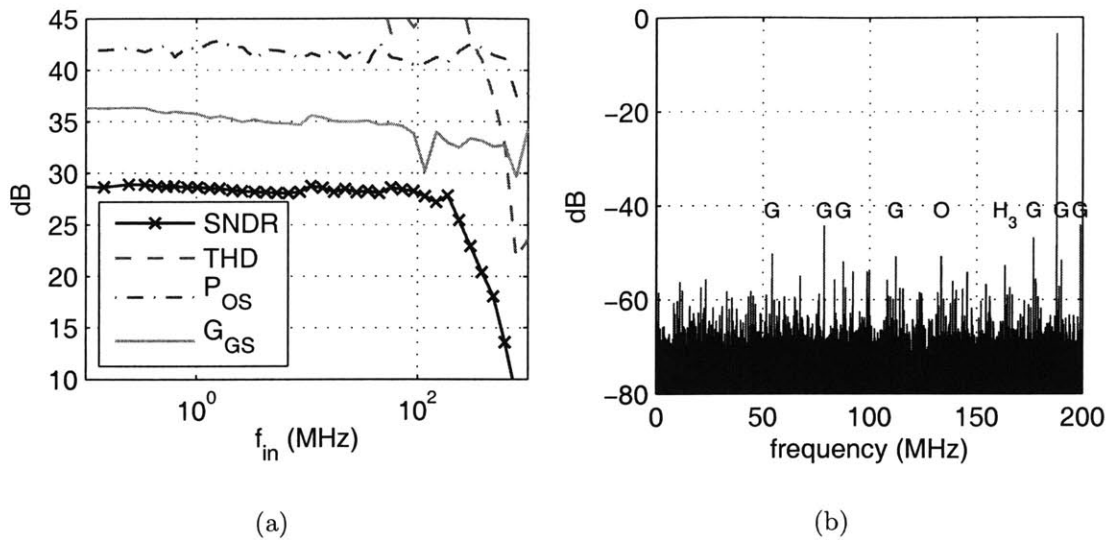


Figure 6-27: Dynamic performance of the ADC at 400 MS/s, showing (a) the SNDR versus input frequency and (b) an FFT of a 190 MHz input.

Table 6.2: Power breakdown of ADC

Supply	Description	Voltage	Current	Voltage	Current
		$f_s = 250 \text{ MS/s}$		$f_s = 400 \text{ MS/s}$	
$V_{DDD}$	Core digital	800 mV	$720 \mu\text{A}$	1.0 V	1.39 mA
$V_{DDA}$	Core analog	800 mV	$300 \mu\text{A}$	1.0 V	$600 \mu\text{A}$
$V_{DDC}$	Clock	800 mV	$145 \mu\text{A}$	1.0 V	$250 \mu\text{A}$
$V_{DDS}$	Sampling	1.2 V	$200 \mu\text{A}$	1.2 V	$310 \mu\text{A}$
$V_{REF}$	Reference	350 mV	$89 \mu\text{A}$	350 mV	$140 \mu\text{A}$

the Nyquist ENOB to 3.96. These output mux failures can produce outputs that are off by several bits and therefore the ADC's BER [71] increases significantly.

The per-channel preamplifier biasing has also been tested, as shown in Fig. 6-28. As  $PREI$ , the digital code to the DAC, increases, the gains of the preamplifiers also increase, reducing the measured offset voltage. For small  $PREI$ , the channel SNDR degrades because the preamplifiers are effectively attenuating the input below the offset of the latch. As the current gets too large, the PFET preamplifier loads enter saturation, and the input devices enter triode, severely degrading the gain, offset, and SNDR. The preamplifier currents can be chosen on a per-channel basis to meet the overall offset and SNDR specification. Optimizing

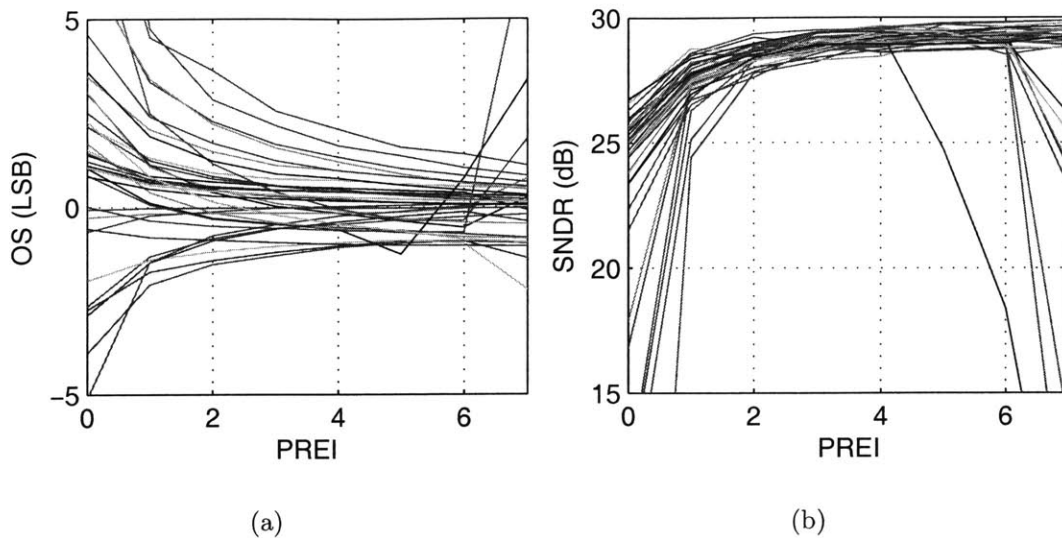


Figure 6-28: Measured channel (a) offsets and (b) SNDR as a function of the programmable preamplifier bias current. Curves for 36 channels are overlaid.

each preamplifier bias current provides current savings at constant overall performance, but the measured savings is only  $80 \mu\text{A}$ . This is roughly 20% of the current drawn from the analog supply but only a small fraction of the overall power. Per-channel power distribution may be more effective for systems that have a larger analog power contribution, but it must be weighed against the additional complexity and test time.

### 6.6.2 Local Variation

Channel selection for redundancy begins with a measurement of the variation across a chip. The raw measurements are converted into the set of metrics INL, DNL,  $\text{SNDR}_L$ ,  $\text{SNDR}_H$ ,  $\text{THD}_H$  offset, gain, timing skew. The subscript “L” denotes low frequency performance, and the subscript “H” denotes Nyquist input. This input frequency is based on the overall interleaved sampling rate, not the sampling rate of an individual channel. This set of parameters requires only two measurements per channel, one with a low frequency sinusoidal input, and one with a high frequency input. The interleaved outputs of the ADC include measurements from 36 channels at the same time, and only four total measurements are needed to generate

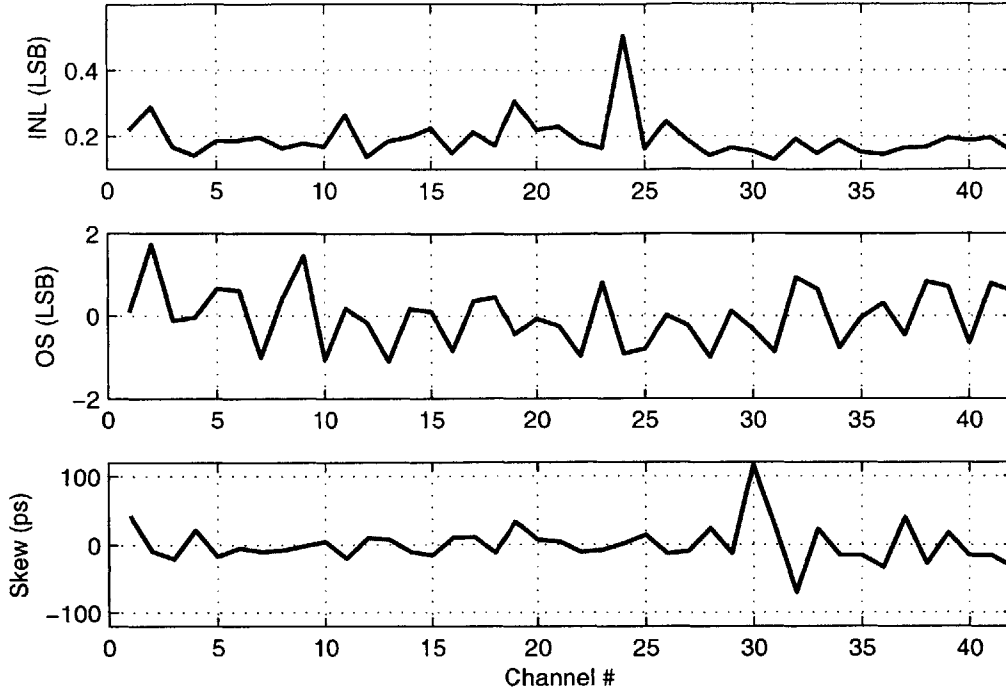


Figure 6-29: Variation of INL, offset, and skew across the 42 channels on a die.

all the data: two for channels 1–36, and two for channels 1–30 and 37–42.

With the exception of timing skew, all of the parameters can be based on looking at the outputs of a single channel. Timing skew is extracted from the Nyquist input test by individually sine fitting the output of each channel. The channel phases are unfolded to a monotonically increasing sequence of phases that are then converted into time delays with the known input frequency ( $\Delta t = \Delta\phi / (2\pi f_{in})$ ). Timing skew is the variation between the measured and ideal channel sampling instants. Without an accurate timing base, the ideal sampling instant is set such that the average timing skew of the first 30 channels measured in that run is 0.

Figure 6-29 shows a typical variation of some of these metrics across the 42 channels on a single chip. Unless explicitly mentioned otherwise, all channel variation and redundancy measurements are performed at  $V_{DD} = 0.9\text{ V}$  and  $f_s = 250\text{ MS/s}$ . The total level of variation does limit the yield, but the structure of the variation is critical to the yield improvement achievable through redundancy. Redundancy is most effective when correcting

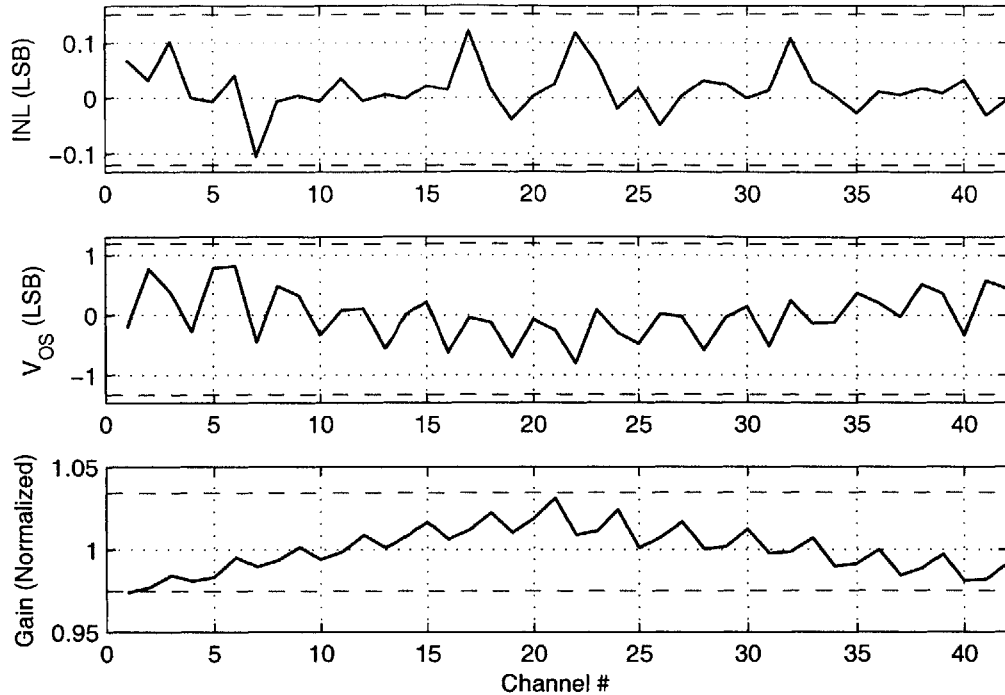


Figure 6-30: Average of the static parameters (INL, offset, gain) for each channel across all measured dies. The dashed lines indicate the 5th and 95th percentiles of the parameters cross all measured channels. The INL measurement at top is the INL of the MSB transition, not the overall INL for the channel. The measurements on all chips were performed at 250 MS/s with a 0.9 V core voltage.

“white” variations, where every channel is completely independent. If the same channels are bad on every chip (i.e., systematic variation), it is an implementation-induced error that can be corrected through improved layout. Alternatively, if the bad channels are clustered within a single block (e.g., if the block sampling network is defective), then the two redundant channels will be insufficient to correct all the errors.

Systematic variation is determined by averaging the parameters of each channel across all the dies, as shown in Figures 6-30 and 6-31. As only 24 chips were available for measurements, this is only an estimate of systematic variation, and this same caveat applies to any other general conclusions about variation and yield. Where present, the dashed lines indicate the 5th and 95th percentiles across all channels. The redundant channel selection procedure presented in Section 6.6.3 typically replaces channels that fall outside of these bounds. In

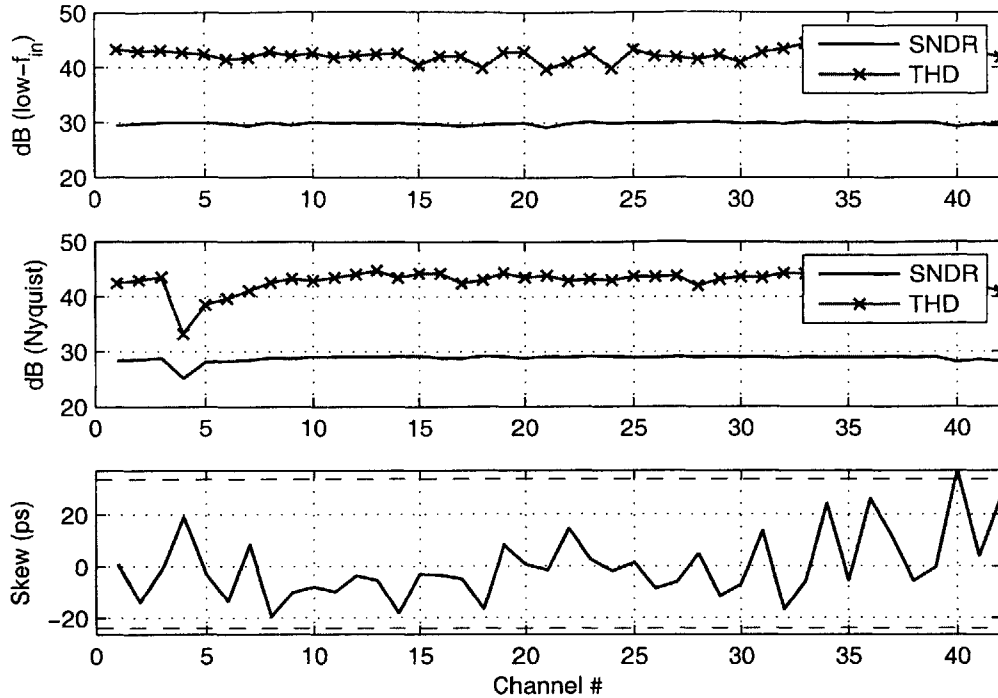


Figure 6-31: Average of the dynamic parameters for each channel across all measured dies. With the exception of channel 3 at the high input frequency, the SNDR of all channels exhibits no systematic variation, as is it dominated by quantization noise and, at high frequencies, jitter.

general, the level of systematic variation is less than the level of random variation. A few of these plots, however, require further consideration.

The offset voltage displays a clear structure where channels 1,4,...,40 have a more negative offset voltage than the rest of the channels. This is a result of systematic block variation, as all of these channels are in block 1. The first block has an average offset voltage 0.64 LSB less than the other two blocks. While the layouts of blocks 2 and 3 are identical, the first block has an added circuit to generate the *START* signals. This circuit asymmetrically couples to the differential switched ground nodes in the hierarchical sampling network, causing the systematic block offset.

The gains show a significant bowing, with increased gain for the channels in the middle that are located towards the top of the die. The reference voltage supply is routed from the bottom of the die, and the increased  $IR$  drop seen by the upper channels attenuates  $V_{REF}$ ,

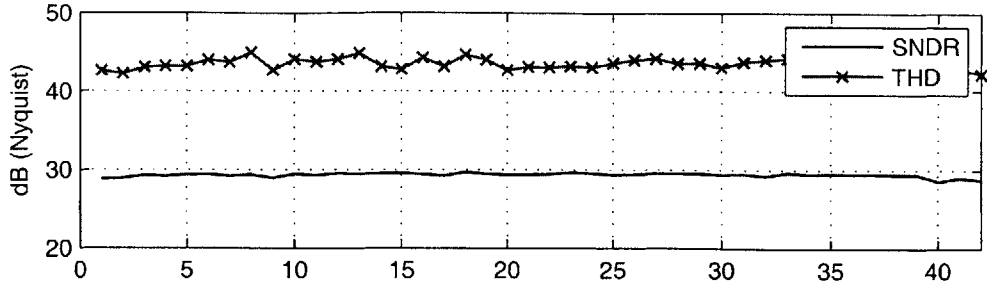


Figure 6-32: Systematic variation of SNDR and THD with the debug bus turned off. As shown, the SNDR drop in channel 4 is eliminated.

which is equivalent to an amplification of the input signal.

While the SNDR and THD for low frequency inputs shows almost no variation, channel 4, the second channel in block 1, shows clearly degraded THD and SNDR with a Nyquist input. While not all chips exhibit this degradation, enough do that channel 4 must be replaced with a redundant channel on a disproportionate number of chips. This error is due to coupling from the first block’s debug bus onto the input signal. The data on the debug bus transitions at the sampling instant of the 4th channel. As the amount of switching activity on this bus is related to the input frequency, it has greater impact with a Nyquist input than a near-DC input. The source of the error has been confirmed by retaking all of the measurements with the debug bus disabled, as shown in Fig. 6-32. While in this particular case, the systematic error is preventable at test time and certainly could be eliminated during the design and layout, redundancy can effectively correct for this type of systematic error because it is isolated to a single channel.

Channel redundancy is particularly ineffective at correcting variations confined within a single block or within the global clock and mux networks. Variation within a block is estimated by averaging the parameters of all the channels within a block and is dominated by variation in the master sampling network. Of particular interest is the timing skew, which motivated the use of the hierarchical sampling network. At the nominal voltage, the standard deviation of the timing skew across the channels is  $22 \text{ ps}_{\text{rms}}$ . According to (6.8), this produces a maximum of 4.6 ENOB for a 250 MHz input, although redundancy can be used to improve the timing skew on-chip, as described below. The skew in delivering the



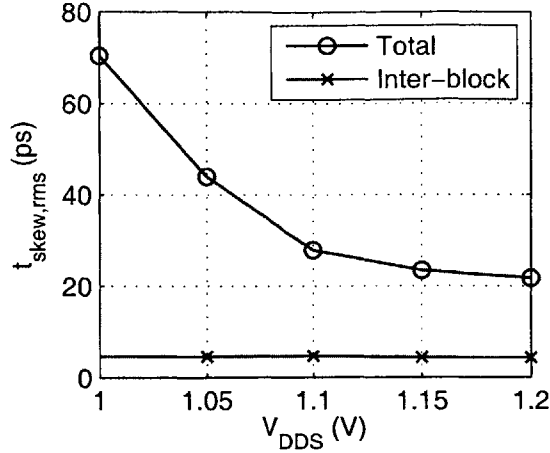


Figure 6-33: Dependence of the timing skew on the sampling voltage  $V_{DDS}$ . The overall skew improves at higher voltages because of reduced delay variation of the signals within the block, but the skew between blocks remains unchanged.

clock to the three master sampling switches is measured by finding the average skew to the channels within each block, and this inter-block skew is  $5.9 \text{ ps}_{\text{rms}}$ . The rest of the skew can be attributed to the intra-block skew, particularly the delay variation of the turn off signals of the per-channel slave switches. The voltage dependence of the skew is plotted in Fig. 6-33. As  $V_{DDS}$ , which drives the per-channel slave switches, increases, the skew decreases as expected. The inter-block skew, dominated by routing variations, is relatively insensitive to voltage.

Table 6.3 summarizes the variation data. The total variation among all  $24 \times 42$  channels tested is split into systematic variation (correlated variations among the same channels on different chips), global variation (correlated among all the channels on a single chip), block-correlated variation, and, finally, local channel variation. The values in the table are explicitly calculated as follows. On die  $d$ , within block  $b$ , channel  $c$ 's metric  $Z$  (e.g., INL) is considered as the sum of separate error sources:

$$Z(d, b, c) = Z_{\text{systematic}}(b, c) + Z_{\text{global}}(d) + Z_{\text{block}}(d, b) + Z_{\text{channel}}(d, b, c) \quad (6.15)$$

The separate error sources are approximated from the measured metrics  $Z(d, b, c)$  as shown

Table 6.3: Summary of channel variations

	Total $\sigma$	Systematic	Global	Inter-block	Channel
INL (LSB)	0.184	0.042	0.001	0.001	0.170
DNL (LSB)	0.169	0.049	0.010	0.002	0.153
$V_{OS}$ (LSB)	0.764	0.409	0.004	0.007	0.589
Gain	0.019	0.013	0	0	0.004
SNDR @ low- $f_{in}$ (dB) <sup>a</sup>	3.707	0.293	0.05	0.004	1.045
SNDR @ Nyquist (dB) <sup>a</sup>	1.189	0.692	0.005	0.021	0.893
THD @ Nyquist (dB) <sup>a</sup>	3.46	2.065	0.443	0.119	2.564
$t_{skew}$ (ps)	23.609	15.793	0 <sup>b</sup>	1.335	16.022

<sup>a</sup>Variance of the log quantity

<sup>b</sup>Average skew on a die is 0 by definition because of the lack of an absolute timing reference.

in (6.16)–(6.19). The values in Table 6.3 are the standard deviations of these separate error sources. Since the distributions of these metrics are not Gaussian or independent, there is no direct way to get the first column (total variance) through simple arithmetic manipulation of the last four columns.

$$Z_{systematic}(b, c) = \frac{1}{N_{chip}} \sum_{d=1}^{N_{chip}} Z(d, b, c) \quad (6.16)$$

$$Z_{global}(d) = \frac{1}{M_{total}} \sum_{b=1}^3 \sum_{c=1}^{12} [Z(d, b, c) - Z_{systematic}(b, c)] \quad (6.17)$$

$$Z_{block}(d, b) = \frac{1}{12} \sum_{c=1}^{12} [Z(d, b, c) - Z_{systematic}(b, c) - Z_{global}(d)] \quad (6.18)$$

$$Z_{local}(d, b, c) = Z(d, b, c) - Z_{systematic}(b, c) - Z_{global}(d) - Z_{block}(d, b) \quad (6.19)$$

Except for gain, random channel variation has the largest contribution to the overall variation of every parameter. The gain variation is dominated by the  $IR$  drop on the  $V_{REF}$  line discussed earlier. Global and inter-block variation, both of which cannot be corrected by the implemented redundancy scheme, are relatively minor.

### 6.6.3 Redundant Channel Selection and Yield

Once all of the channels on a die have been characterized, the “best” 36 are selected. What constitutes the best channels is a matter of interpretation and depends on the target specifications. This section discusses selection strategies and proposes a cost function for automated channel selection. Overall yield improvement is presented as the number of redundant channels per block is increased from 0 to 2.

First, channel selection is examined through an example, using the measured variation plotted in Fig. 6-29, which shows relatively constrained distributions with a few outliers. Channel 24 has the largest INL, channels 2 and 9 have the peak offsets, and channels 30 and 32 exhibit the worst timing skews. If global redundancy had been implemented, these five channels could be replaced, improving the worst-case INL ( $0.50 \rightarrow 0.31$  LSB), peak-to-peak offset ( $2.8 \rightarrow 2.0$  LSB), and rms timing skew ( $28.5 \rightarrow 18.8$  ps) by more than 30% at the same time, with one channel left over. With the implemented block redundancy, however, channels 9, 24, and 30 are all located in the third block and cannot be replaced simultaneously. Keeping channel 9 gives the same improvement in INL and skew but the peak-to-peak offset is improved to only 2.5 LSB.

Before defining a strategy for automated channel selection, the effect of each of the metrics must be considered. INL gives an indication of the worst case error for a slowly varying input signal, whereas THD is the best characterization of the linearity error of a high frequency input. As offset is corrected digitally, the analog offset,  $V_{OS}$ , only limits the peak signal before clipping occurs. The largest absolute offset voltage of the active channels limits the peak input signal; the offsets of the other active channels have no effect. Gain shows little random variation and is not used as one of the inputs for channel selection. The overall SNDR has contributions from every channels' SNDR plus the degradation of any of the mismatch effects. While the low frequency SNDR does exhibit variation, it is highly correlated with the offset due to the measurement setup where channels with the largest offsets will clip the -1 dBFS input signal; it is therefore ignored for this selection procedure. At high input frequencies, SNDR is often correlated with THD (cf. channel 4 in Fig. 6-

31); severe high frequency SNDR degradation that is uncorrelated with THD is usually a sign of output mux failure and can lead to significant degradation of overall performance. Finally, timing skew can lead to voltage errors for high frequency input and a corresponding degradation in overall SNDR according to (6.8).

The selection process proposed here first eliminates the channels that are outliers of global criteria. Then, if too many channels remain, the worst channels on the chip, relative only to each other, are removed. For the global specification selection, every channel  $i$  is assigned a total penalty  $P_i$  that is the sum of penalties from the individual parameters of that channel (6.20). Each individual penalty is nonnegative, and the parameters  $k_j$  weight the penalties to normalize the different scales and impact of the penalties.

$$P_i = \sum_j k_j p_{j,i} \quad j \in \{INL, V_{OS}, SNDR_H, THD, t_{skew}\} \quad (6.20)$$

The individual penalties are calculated as in (6.21)–(6.25), where  $(x)^*$  is  $x$  for  $x > 0$ , and 0 otherwise.

$$p_{INL,i} = (INL_i - T_{INL})^* \quad (6.21)$$

$$p_{V_{OS},i} = \begin{cases} (|V_{OS,i}| - T_{V_{OS}})^* & |V_{OS,i}| = \max_k |V_{OS,k}| \\ 0 & \text{o.w.} \end{cases} \quad (6.22)$$

$$p_{SNDR_H,i} = (T_{SNDR_H} - SNDR_{H,i})^* \quad (6.23)$$

$$p_{THD,i} = (T_{THD} - |THD_i|)^* \quad (6.24)$$

$$p_{t_{skew},i} = (|t_{skew,i}| - T_{t_{skew}})^* \quad (6.25)$$

For the global channel selection, the targets  $T_{INL}$ ,  $T_{V_{OS}}$ ,  $T_{SNDR_H}$ ,  $T_{THD}$ , and  $T_{t_{skew}}$  are fixed for all chips. The offset penalty (6.22) is nonzero only for the channel with the largest absolute offset of the non-eliminated channels. Initially, the penalties of all 42 channels are calculated, and then the channel with the largest penalty is eliminated. The penalties are then recalculated for the remaining 41 channels (only  $p_{V_{OS},i}$  might change), and the

Table 6.4: Weighting coefficients and target parameters for channel selection

Weighting Coefficient		Target	
$k_{INL}$	1	$T_{INL}$	0.4 LSB
$k_{VOS}$	0.5	$T_{VOS}$	1 LSB
$k_{SNDR_H}$	3	$T_{SNDR_H}$	28 dB
$k_{THD}$	1	$T_{THD}$	36 dB
$k_{t_{skew}}$	$(1/30) \text{ ps}^{-1}$	$T_{t_{skew}}$	30 ps

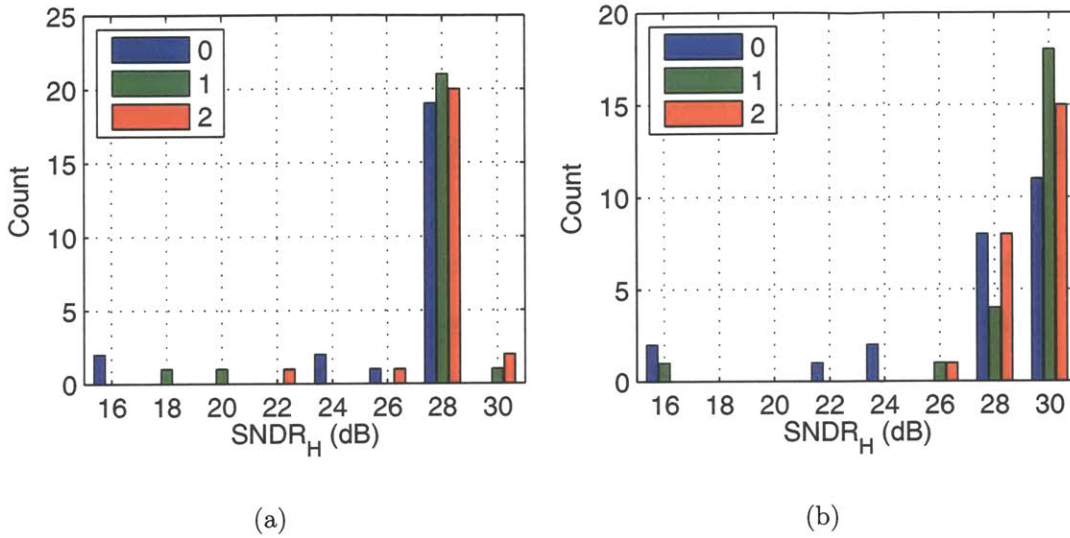


Figure 6-34: Histogram showing yield improvement as a function of the number of redundant channels per block for the ADC operating at (a) 250 MS/s and (b) 125 MS/s.

remaining channel with largest penalty is eliminated. No more than two channels per block can be eliminated. This process terminates when each block has two channels eliminated, or no channels with nonzero penalties remain for blocks that have fewer than two channels eliminated. If the latter case holds, the process is repeated but with the global targets  $T_j$  replaced with the average of those parameters across the remaining channels. Although this has not happened on any of the tested chips, it is conceivable that a block will still not have eliminated 2 channels after this second step. This corresponds to the case where the channels in this block are all very good, exceeding the averages of the other blocks. In this case, the targets can be re-evaluated to the averages of this block, or channels can be eliminated at random. The remaining 12 channels per block are chosen for operation.

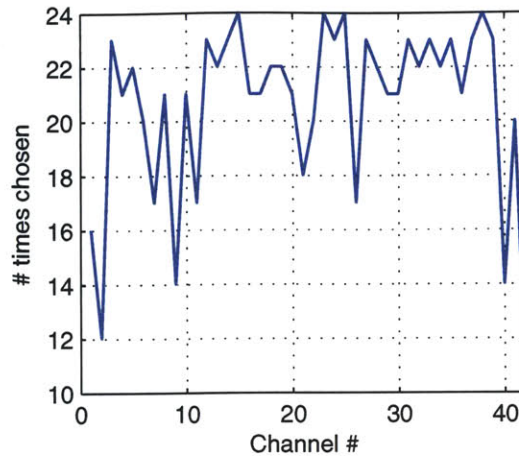


Figure 6-35: Number of times the given channel was chosen as “good” out of the 24 dies.

This selection process is implemented using the weighting and target parameters in Table 6.4. The high frequency SNDR is given the greatest weight as measurements have demonstrated that it has the greatest impact on overall SNDR, and it is a good indicator for a range of failures (e.g., noise, digital failures) that are not well captured by the other metrics. Figure 6-34 shows the worst case  $\text{SNDR}_H$  of active channels as a function of the number of redundant channels per block. The number of redundant channels is varied by the selection mechanism by forcing the last two or one channels to be eliminated first to mimic the effect of zero or only one redundant channel per block. The number of chips with a worst case SNDR of less than 28 dB decreases from 5 to 2 (Fig. 6-34(a)). Also shown is the improvement at a slower sampling frequency of 125 MHz (Fig. 6-34(b)), which is measured at 125 MS/s, where practically all timing related non-idealities are eliminated.

The results of the selection process can be compared with the systematic variation data presented earlier by plotting the number of times a channel was selected (or, equivalently, eliminated) over the 24 tested dies, as shown in Fig. 6-35. The worst channels are those selected the fewest number of times. Channels 40 and 42 have the worst systematic timing skews (Fig. 6-31), and channel 2 tends to have a higher INL and offset. The systematic variation serves as a predisposition for a bad channel, but since none of the channels were eliminated for every chip, random local variation can trump systematic variation for channel

Table 6.5: Yielding chips (out of 24) as a function of the number of redundant channels

Metric	Specification	Redundant Channels/block		
		0	1	2
INL	$\leq 0.6$ LSB	14	21	21
$V_{OS}$	$\leq 2$ LSB	19	23	24
$SNDR_H$	$\geq 27$ dB	19	22	22
$THD$	$\geq 33$ dB	21	23	24
$t_{skew}$ (rms)	$\leq 15$ ps	24	24	24
$SNDR_H^a$	$\geq 27$ dB	17	23	23
All specs		10	21	21

<sup>a</sup>SNDR of interleaved ADC

selection.

A summary of the improvement in these parameters with redundancy is shown in Table 6.5. For best results, the target for channel selection (Table 6.4) should be at least as conservative as the specifications listed in this table. Also included in the table is the yield of the overall SNDR for the interleaved chip, which also shows improvement with redundancy.

#### 6.6.4 BIST Extension

The channel selection procedure described in this chapter is offline and suitable for initial die testing but is not directly useful for dynamic channel selection after system deployment. For some applications, it is desirable to change the active channels in response to changing environmental conditions, such as voltage, temperature, or in response to process aging. To highlight the usefulness of BIST, Figure 6-36 presents the correlation of selected channels on a chip for six different temperature settings from 0 to 80 °C. While channels 36 and 42, with the highest timing skews, are least used, the dominant failure mechanisms change at high temperatures from skew to linearity, meaning that no channel is the worst across the entire temperature range. This section briefly describes BIST procedures that could lead to automatic, dynamic channel selection. A BIST operates at a very low duty cycle, and its energy overhead during normal operation would be confined to leakage power only.

The actual channel selection procedure (6.20)–(6.25) is implementable using a finite state

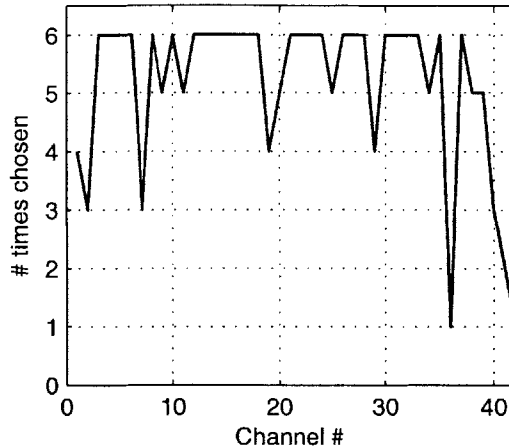


Figure 6-36: Number of times the given channel was selected for active operation, where measurements were performed on a single die at 6 different temperature settings from 0 to 80 °C. While some channels performed well across all temperatures, no channel was the worst at every setting.

machine (FSM) and simple adder/comparator. The difficult part is measuring the performance metrics directly on-chip. One advantage of having redundant channels is that foreground characterization/calibration of the channels can be done while still leaving 36 active channels. This is similar to the approach proposed in [47] where an extra channel is included specifically to allow foreground calibration of channels without disturbing overall ADC performance. There may be some performance degradation because the “best” 36 channels cannot be selected while channel calibration is being performed, but 36 channels are always available.

The static linearity and offset can be directly measured using the SAR calibration procedure in [31], where the calibration engine and resistive sub-DAC can be shared among all channels. Gain errors can be measured using the technique in [47].

Dynamic errors are more difficult to measure. A signal generator can be included on-chip for foreground calibration/characterization. Another possibility is to measure the channels with the actual input signal. Jamal, et al., proposed adding chopping switches at the input of each channel, thereby creating a mechanism to calibrate offset, gain, and timing skew errors with arbitrary input signals [53].



Table 6.6: Performance summary of 36+6-way interleaved ADC

Technology	65-nm CMOS 1P6M	
Resolution	5 bit	
Input Range	700 mV <sub>pp</sub> Differential	
Sampling Rate	250 MS/s	400 MS/s
Core Supply Voltage	800 mV	1.0 V
Sampling Supply Voltage	1.2 V	1.2 V
SNDR (DC)	29.2 dB	28.8 dB
Input frequency	117 MHz	188 MHz
SNDR	28.4 dB	27.8 dB
SFDR	47.5 dB	43.7 dB
THD	-44.6 dB	-45.7 dB
Analog Power	270 $\mu$ W	650 $\mu$ W
Digital Power	930 $\mu$ W	2.01 mW
Total Power	1.20 mW	2.66 mW
Figure of Merit (1.1)	240 fJ/conv. step	350 fJ/conv. step
Active Area	2.10 mm $\times$ 1.45 mm	

Another method to measure dynamic errors is to directly compare the outputs of multiple channels sampling at the same time. With two redundant channels per block, up to three channels can sample the same input (assuming an appropriate modification of the channel synchronization circuitry, Fig. 6-7), and directly comparing the outputs gives the relative errors of the channels. This method can be combined with chopping to give meaningful dynamic results even for stationary inputs. Sampling multiple channels at the same time does introduce extra potential for coupling errors, which must be considered when determining the effectiveness of this scheme.

## 6.7 Chip Summary

This chapter has presented the design of a robust highly-parallel ADC designed to operate near the minimum mixed-signal energy point. The ADC is composed of three blocks with fourteen channels each, two of which are redundant. Clock distribution power is kept in check by distributing a single low frequency/voltage clock to the channels within a block; only one

sampling switch per block receives the precise full speed sampling clock. This master switch defines the sampling instant per block, reducing timing skew, a pernicious problem to correct digitally. Finally, redundant channels counteract yield loss due to local variations in parallel circuits. The redundancy implementation requires minimal overhead and allows for a highly energy efficient final design. The ADC performance is summarized in Table 6.6.

# Chapter 7

## Conclusion

### 7.1 Summary of Contributions

High-data-rate UWB radio transceivers require high-speed low-resolution ADCs. Three 5-bit time-interleaved SAR ADCs have been implemented with an emphasis on energy efficiency. While traditionally limited to low-speed applications, this work has furthered the emergence of the SAR architecture for high-speed applications. Techniques for joint analog and digital design have been explored, and circuits for highly-interleaved ADCs have been designed. Parallelism is explicitly used to improve energy efficiency in mixed-signal circuits without sacrificing yield.

The first chip, implemented in a 0.18- $\mu\text{m}$  CMOS process, featured a custom digital logic controller that was necessary to achieve the channel sampling rates at reasonable power consumptions and included a duty-cycled comparator and self-timed bit-cycling. The second testchip is the first published ADC in 65-nm CMOS. Self-timed bit-cycling has been replaced with a separate timing adjustment that uses a VDL to statically pre-borrow settling time from the latch to the preamplifier.

The split capacitor array has been introduced as a replacement for the conventional binary weighted capacitor array DAC in a SAR ADC. The split array retains the insensitivity of its linearity to stray capacitance between any node and ground. This novel structure splits the

MSB capacitor into an identical copy of the rest of the array; it does not require an increase in the total capacitance or area of the array. Then, during every bit decision after the first, only one capacitor from the MSB or main subarray is switched, reducing the total switching energy by 37%. In addition, the total current drawn for a conversion is made symmetric with the input signal, removing any even order harmonics from the input-signal dependent current drawn from a reference voltage buffer. Finally, correlations in unit capacitors for neighboring bit decisions reduce the DNL by  $\sqrt{2}$  with the same INL.

To address the total power consumption of an ADC, a mixed-signal energy model has been developed that can find the genuine optimum energy point, considering both the analog and digital circuits and their interactions. The model's framework has been designed to allow flexibility in analyzing tradeoffs across the analog/digital boundary; a coupled behavioral model shows the response of the ADC to non-idealities, variations, and digital correction schemes. The optimization was performed to find the best combination of operating voltages and number of parallel channels, which is limited by clock and input buffer

While the optimum number of channels is 60, the minimum is sufficiently broad that 36 channels suffers less than a 10% energy penalty and was chosen as the level of parallelism in the final testchip. This chip is the first demonstration of the use of parallelism in mixed-signal circuits specifically to drive down the voltage for energy savings. Offset mismatch between channels is corrected digitally. Timing skew is minimized using a hierarchical top-plate sampling network that has been extended to permit overlapped sampling periods with a slow turn-on input switch drive for reduced crosstalk between channels. The channels have been partitioned into three blocks, and the clock distribution has a precise domain connected only to the master sampling networks and an imprecise domain, at lower voltages, distributed to the channels. Finally, this work is the first implementation of channel redundancy to combat yield loss in parallel ADCs.

A comparison of the three ADCs presented within this thesis with state-of-the-art high-speed low-resolution converters is shown in Fig. 7-1 and Table 7.1, using the figure of merit defined in (1.1) for comparison. The final two testchips show performance similar to the

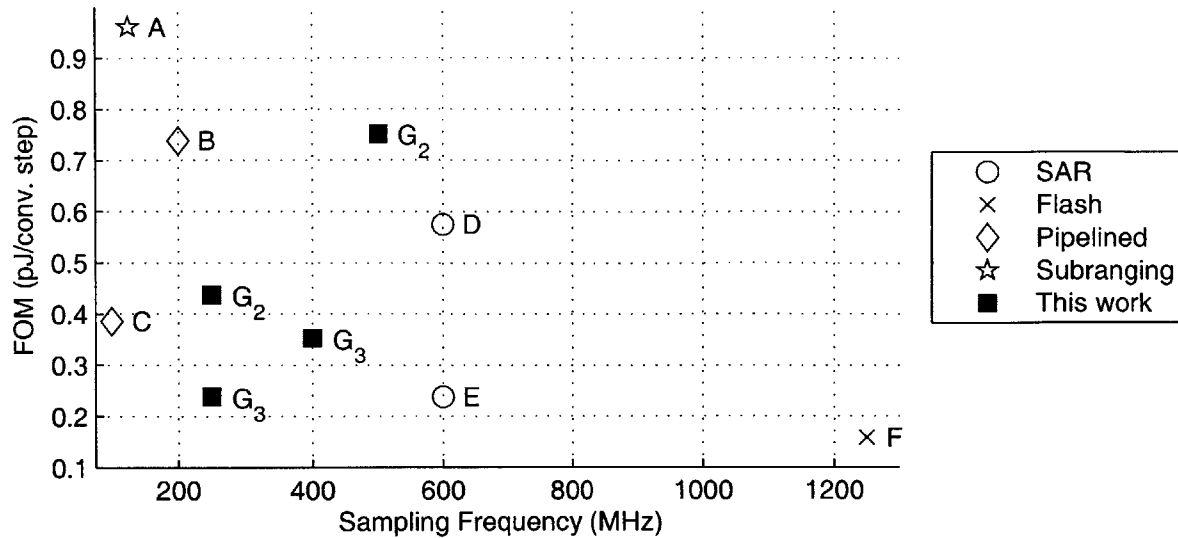


Figure 7-1: Comparison of energy efficiency figure of merit for state-of-the-art converters with a resolution  $\leq 8$  bits and sampling rates exceeding 100 MHz.

best published converters to date. In addition, the SAR ADC is clearly shown as being competitive with the more traditional high-speed architectures.

## 7.2 Conclusions

A unifying theme throughout this thesis has been the use of parallelism specifically to improve the energy efficiency of mixed-signal circuits, in contrast to its conventional role of only increasing maximum operating speeds. Time-interleaving allows the energy efficient slow SAR architecture to operate at speeds comparable to non-interleaved flash converters but at reduced power. Furthermore, increased levels of parallelism slow individual channel operation and permit lower voltage supplies and additional improvement in energy efficiency. While the beneficial relationship between parallelism and energy is not new [94], it has been mainly confined to digital circuits.

Perhaps it is not surprising, because of the significant digital power consumption in the three chips, but many of the ideas in this thesis are similar to or inspired by digital circuit techniques. Clearly, the dynamic logic controller is a direct implementation of high-

Table 7.1: Comparison of state-of-the-art ADCs.

	Work	Architecture	Feature Size	Power (mW)	$f_S$ (MHz)	Resolution (bits)	$f_{in}$ (MHz)	ENOB	FOM (pJ/conv. step)
A	[133]	Subranging	0.13 $\mu\text{m}$	21	125	8	62.5	7.5	0.96
B	[23]	Pipelined	0.18 $\mu\text{m}$	30	200	8	99	7.68	0.74
C	[27]	Pipelined	0.18 $\mu\text{m}$	4.6	100	8	50	6.9	0.39
D	[46]	SAR	90 nm	10	600	6	300	5.1	0.5
E	[45]	SAR	0.13 $\mu\text{m}$	5.3	600	6	300	5.02	0.27
F	[134]	Flash	90 nm	2.5	1250	4	625	3.66	0.16
G <sub>1</sub>	This thesis	SAR	0.18 $\mu\text{m}$	7.8	500	5	20	3.06	23
G <sub>2</sub>			65 nm	5.9	500	5	239	4.04	0.75
				1.8	250	5	120	4.10	0.44
G <sub>3</sub>			65 nm	1.2	250	5	117	4.42	0.24
				2.7	400	5	188	4.32	0.35

performance digital circuits. Less obviously, the split capacitor array is partially motivated by charge conservation in bus switching, and self-timed bit-cycling is not unlike asynchronous or self-timed digital circuitry. Finally, redundancy (parallelism to remedy the impairments of parallelism) is standard practice in digital memories. High-speed ADC designers must continue to draw from the best of high-speed and low-power digital techniques, and combine it with the state-of-the-art analog design.

When faced with the challenges in designing analog circuits in deep sub-micron CMOS, architectural choices make a great impact on design difficulty. The reemergence of active research on the SAR topology should pay great dividends, and while it is unlikely to supplant pipelined as the dominant high-speed architecture, pipelined converters in advanced technologies may be dissimilar to the traditional operational amplifier based designs of present. The guidelines for design in 65-nm CMOS are to keep analog complexity low, avoid any high gain linear stages, and minimize dependence on precise matching between active devices. Redundancy is a more effective way to correct for static impairments (i.e., linearity, offset, and skew) than over-designing for precise matching in the first place.

For parallel systems, redundancy is an incredibly powerful tool to add to calibration for yield improvement and/or relaxed channel specifications. The key is to design a flexible

enough architecture that the overhead for enabling and disabling parts is small (in the final testchip, it was only a few gates per channel). By using redundancy at a higher level of the hierarchy, more faults can be corrected, particularly if fault models are not known a priori. Unfortunately, redundancy is not a panacea: it cannot correct for errors above the level where redundancy is applied (e.g., the output mux failure at the block level), but significant yield improvement is achievable by applying redundancy and calibration at multiple levels of the hierarchy, and reserving the most conservative design for the non-correctable portions of the circuit.

### 7.3 Future Directions

While the mixed-signal energy optimization was set up for a SAR ADC, the design methodology is directly applicable to other ADC topologies, many of which have significant digital power (cf. Fig. 5-1) and opportunities for moving analog complexity to the digital domain. In addition, many larger systems, from radios to sensor interfaces, span the analog/digital boundary and can also be optimized using such a model. A single tool that can integrate the behavioral model and energy model could make this optimization methodology more practical. In addition, different levels of abstraction should be added to this model, from transistor level to digital synthesis. Ultimately, such a tool could lead to automated design of mixed-signal circuits.

Redundancy is still an underused technique, particularly in the analog and mixed-signal domains. This work showed the advantage of redundancy over sizing for yield improvement in parallel circuits, but additional research is necessary to explore the comparative advantages of redundancy and calibration. If the overhead of redundancy is sufficiently small, it may have an energy advantage over calibration or other correction methods because, once the “bad” blocks have been eliminated, the remaining “good” blocks operate without the added correction complexity and power. For circuits that are not inherently parallel, redundancy may still be applicable. For instance, as shown in the split ADC (not to be confused with the split capacitor array), a high accuracy block can be split into two lower accuracy blocks

without an overall increase in power or decrease in resolution [135]. This introduction of parallelism was used to speed up background calibration, but redundancy could be applied to the lower accuracy sub-blocks, improving the overall yield. Redundancy is certainly not a panacea, as it cannot get around certain circuit limitations, such as noise, but its potential has certainly not been fully explored.

For all of these yield improvement techniques, there needs to be a better understanding of variation and fault models. While significant data exists on transistor or capacitor variation and mismatch, easily incorporating this information into large mixed-signal circuits remains a challenge. Monte Carlo simulations are simply not practical for a large system, particularly when local variation requires statistical models of thousands of elements varying independently. For this thesis, Monte Carlo simulations were done only for very small circuit blocks (e.g., comparator, sampling switch, etc.), and the variations of these blocks were considered independent. Unfortunately, not every block, such as the clock distribution, or SAR digital controller were able to be fully simulated for local variation, and therefore no comprehensive fault model could be generated a priori. A simulation or design methodology that can translate the available process variation data into overall fault models of complex circuits would be an invaluable asset, particularly for advanced and still emerging technologies.



# Appendix A

## SAR Behavioral Model

As an aide during the design process, a behavioral model of the SAR ADC is presented in this chapter. During the early design period, the behavioral model is used to determine the response to the ADC to various design tradeoffs. Even after design has begun, the behavioral model is used for variation analysis. For global variation, corner models are often available. These offer a limited number of discrete simulations that must be performed to cover the outer reaches of the process space. Corner models, however, can be too conservative in some cases and do not give predictions about the distribution of the resultant chips. In addition, corner models do not apply to local variation. Thus, Monte Carlo simulations are used to handle statistical variations, but they require hundreds to thousands of runs. A transistor-level simulation of a single conversion of a channel takes on the order of minutes, making Monte Carlo simulations extremely time consuming at best. For longer simulations, such as timing skew between neighboring channels on a long sinusoid, transistor-level Monte Carlo simulations are completely impractical at present. Behavioral simulations offer an alternative to transistor-level simulations, where accuracy of the final results is traded for speed. Several behavioral simulation tools are available [136, 137] that can provide various schematic capture, netlisting, and simulation capabilities. In this work, a behavioral model was implemented directly in Matlab. This appendix describes the general setup of the behavioral model and its capabilities. The results of the behavioral model are used throughout

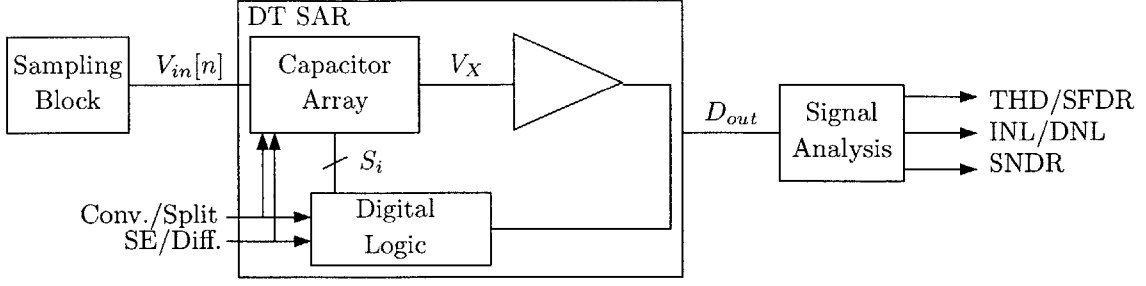


Figure A-1: Block diagram of the behavioral model. The analysis and digital control block are ideal, while the remaining blocks can model non-idealities.

the thesis.

Figure A-1 shows the structure of the behavioral model. At its core is a discrete time SAR ADC that takes sampled analog values as input and produces the corresponding digital outputs. This discrete time SAR includes as sub-blocks a capacitive DAC, a comparator, and digital control logic. The discrete time inputs are generated by the sampling block, which can create arbitrary ramp and sinusoidal inputs. The outputs are processed by an analysis block that computes SNR, SNDR, THD, SFDR, offset, and INL/DNL. All of the blocks behave ideally by default. With the exception of the digital logic and signal analysis block, they also model various non-idealities, as described below. Unless specified below, the positive and negative references are 1 and 0, respectively.

## A.1 Block Descriptions

The DAC can be switched between the binary weighted or split capacitor arrays and be single-ended or differential. Each capacitor within the array includes the sum of i.i.d. errors from its unit capacitors plus a systematic error that models the layout induced errors from nonlinear coupling between top and bottom plate routing. For instance, the MSB capacitor in the conventional array has total capacitance

$$C_{MSB} = \sum_{i=1}^{2^{b-1}} (C_0 + N(0, \sigma_{C_0})) + \Delta C_{MSB}. \quad (\text{A.1})$$

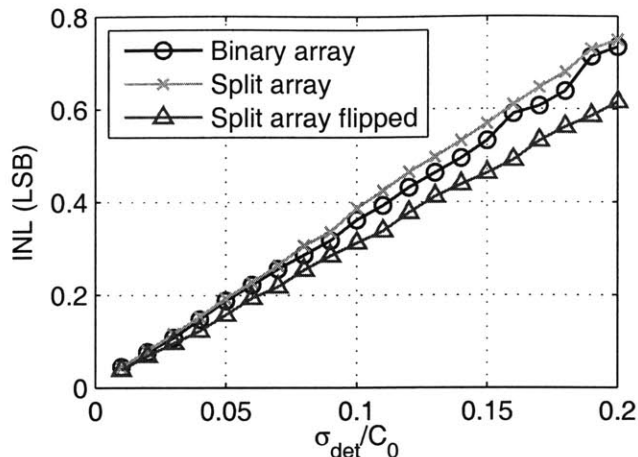


Figure A-2: Behavioral simulation showing the effect of systematic errors of differential capacitor arrays on INL. The model uses  $\Delta C_k \sim N(0, k\sigma_{det})$ . The bottom curve flips the subarrays of the differential split capacitor arrays.

The output of the array is defined by (4.3)–(4.5) where  $C_T$  and  $C_B$  are appropriately modified to include the effect of capacitor variation.

The systematic error can be used to predict the INL and DNL if  $\Delta C_i$  are extracted from the layout. In addition, it can be used to compare the sensitivity of the different arrays' linearities to layout induced errors. Without extracted parasitics, a simplistic error model can be used where, for a capacitor consisting of  $k$  unit capacitors, the systematic error from routing is  $\Delta C_k \sim N(0, k\sigma_{det})$ . This error model is not expected to correspond to most layout induced errors, but without knowledge of the actual layout, this is a reasonable model for early in the design cycle. While it may seem that this definition means that  $\sigma_{C_0}$  in (A.1) can be replaced with  $\sqrt{\sigma_{C_0}^2 + \sigma_{det}^2}$ , that only holds for a single-ended array of a non-time-interleaved converter. Assuming that an identical layout is used for both halves of a differential array and for every channel in an interleaved ADC, then systematic errors are correlated between arrays and channels, whereas the random variation from  $\sigma_{C_0}$  will be independent.

Figure A-2 compares the impact of systematic errors on the linearity of differential ADCs with different capacitor array configurations. The upper two curves are for the binary weighted and split capacitor arrays where both of the arrays in the converter have iden-

tical errors. The lower curve is for an ADC where the two split capacitor arrays have flipped subarrays. For an identical layout of the bottom plate routing within the array, the digital signals to the switches are interchanged between  $C_{5,k}$  and  $C_k$  between the two arrays. This is possible because the two subarrays are symmetric. For this type of systematic error, the overall linearity is improved by flipping the subarrays for the two differential split capacitor arrays.

The ideal comparator outputs a 1 if the input  $V_X > 0$  and 0 otherwise. Non-idealities that can be modeled include the offset  $V_{OS} \sim N(0, \sigma_{OS})$  and finite settling time constant  $\tau$ . The comparator output is

$$COMP[i] = \begin{cases} 1 & V_X[i] + (V_X[i-1] - V_x[i])e^{-1/\tau} > V_{OS} \\ 0 & \text{otherwise,} \end{cases} \quad (\text{A.2})$$

where  $i$  represents the current bit decision period, and  $V_X[0] = 0$  because the comparator is autozeroed during the sampling period at the start of every conversion.

The digital logic generates the appropriate switch signals for the capacitor array using the conventional 1-step switching method [66] or the split capacitor array switch method from Fig. 4-8. No non-idealities are modeled.

The sampling block can pass along arbitrary input signals, or it can generate ramps and sine waves for static and dynamic testing, respectively. While sine waves can be used to determine INL and DNL [85], it is a statistical method and requires many more samples that are unnecessary when an ideal ramp can be generated. The generated sine wave has the form

$$V_{in}[n] = 0.5 + V_{OS} + \frac{A}{2} \sin(2\pi f_{in}(n/f_S + \Delta t_{skew}[n] + \delta t_{jitter}[n])). \quad (\text{A.3})$$

The jitter is assumed to be Gaussian and independent from sample to sample, so that  $\delta t_{jitter}[n] \sim N(0, \sigma_{jitter})$ . The timing skew between channels in an interleaved ADC is modeled by  $\Delta t_{skew}$ , which is periodic with  $M$ , the number of channels.

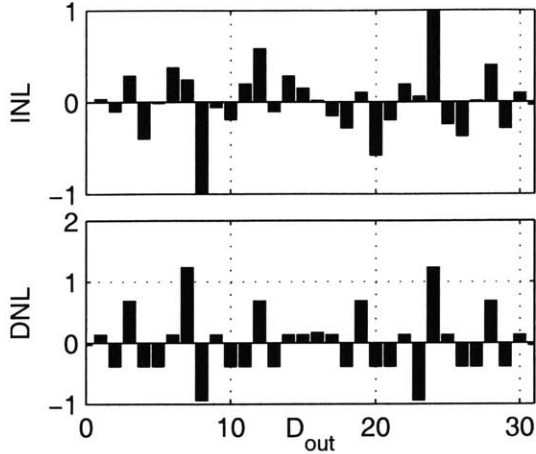


Figure A-3: Static linearity of the ADC with insufficient comparator settling time of  $t_{settle} = 2\tau$ .

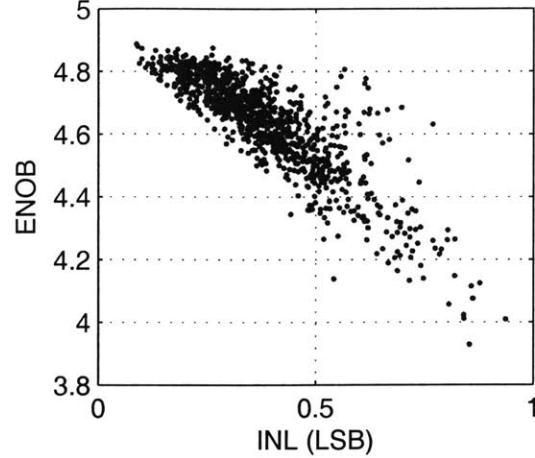


Figure A-4: Scatter plot showing correlation between ENOB and INL under the presence of capacitor mismatch ( $\sigma_{C_0}/C_0 = 0.1$ ).

## A.2 Behavioral Model Results

Several of the results from the behavioral model are included elsewhere (cf. Figs. 2-4, 4-11, 5-6, and 6-2). Presented here is a quick summary of some of the other result highlights. These are included to demonstrate some of the capabilities of the behavioral model. Figure A-3 shows the static linearity when the comparator is too slow. In this case, the comparator is only settling for  $2\tau$ , which leads to severely degraded linearity, particularly at the MSB/2 transitions, when the worst case settling behavior of the comparator is exercised.

Figure A-4 is a scatter plot of the ENOB and INL for the case of only random capacitor errors. In this still relatively ideal case, the INL effectively sets a lower bound on the ENOB; however, when noise and other non-idealities are included, the ENOB will be lower.

Finally, non-ideal sampling is presented in Fig. A-5, which plots the SNDR dependence on the the skew and jitter standard deviations as well as its dependence on input frequency for fixed variation. These results were used to determine the design specifications of the testchips presented in this thesis.

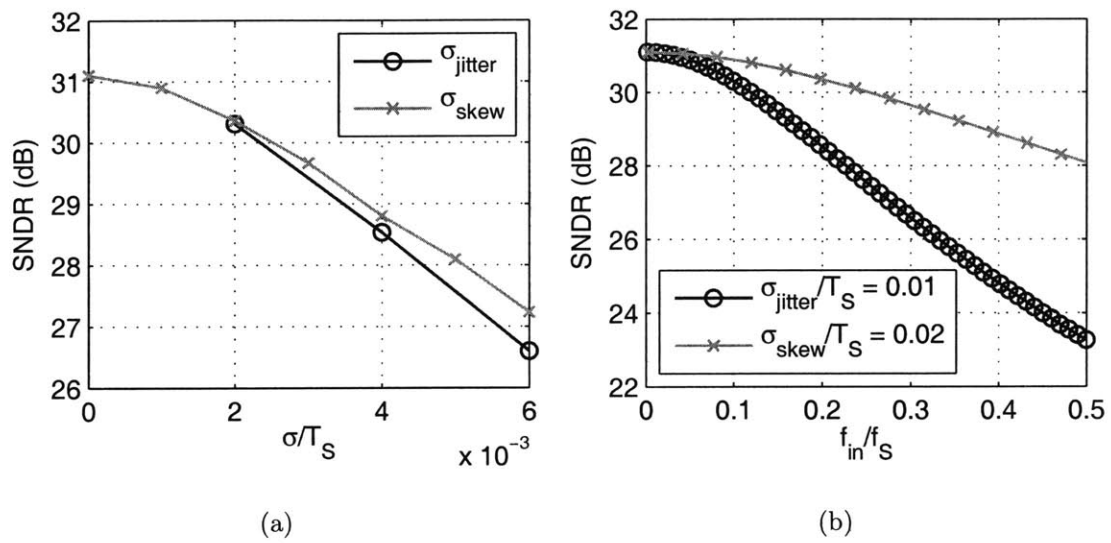


Figure A-5: Behavioral model results showing the degradation in SNDR due to non-ideal sampling. On the left, the degradation due to sampling and skew are compared for a twice Nyquist input. On the right, frequency dependence of the sampling error is plotted.

# Appendix B

## Closed-Form Expression for SAR INL

This section presents the closed-form expression for the INL of a SAR ADC under the case of i.i.d. Gaussian errors in the unit capacitors in the array. In Chapter 4, the variance of the error voltage grows linearly with the output code  $y$  (4.22), but as shown in Fig. 4-11, the INL is maximized for  $y = 2^{b-1}$  and then decreases for larger  $y$ . This analysis begins with (4.20), repeated here for clarity.

$$\frac{V_X(y)}{V_{REF}} = \frac{\sum_{n=1}^b (2^{n-1}C_0 + \delta_n) S_n}{2^b C_0 + \Delta C} = \frac{\sum_{n=1}^b (2^{n-1}C_0 + \delta_n) S_n}{2^b C_0 + \sum_{n=1}^b \delta_n}. \quad (\text{B.1})$$

Here  $\Delta C$  will not be ignored for now. In addition, rather than grouping the errors into the constituent capacitors, the  $\delta_n$  will be expanded into the explicit errors of the unit capacitors,  $\delta_n = \sum_{j=1}^n \epsilon_j$ , where the ordering of  $\epsilon_j$  is ignored, as the specific ordering is only needed for the DNL calculation (cf. (4.25)). Splitting  $\delta_n$  and substituting  $y = \sum_{n=1}^b S_n 2^{n-1}$ , (B.1) can be written as

$$\frac{V_X(y)}{V_{REF}} = \frac{yC_0 + \sum_{j=1}^y \epsilon_j}{2^b C_0 + \sum_{l=1}^b \epsilon_l}. \quad (\text{B.2})$$

The error is the deviation of this voltage from the ideal steps that are placed along a line from 0 to 1, giving

$$\begin{aligned} \frac{V_{err}(y)}{V_{REF}} &= \frac{yC_0 + \sum_{j=1}^y \epsilon_j}{2^b C_0 + \sum_{l=1}^{2^b} \epsilon_l} - \frac{y}{2^b} \\ &= \frac{y2^b C_0 + \sum_{j=1}^y (2^b \epsilon_j) - y2^b C_0 - \sum_{k=1}^{2^b} y\epsilon_k}{2^b (2^b C_0 + \sum_{l=1}^{2^b} \epsilon_l)}. \end{aligned} \quad (B.3)$$

At this point, it is safe to make the approximation that  $\sum_{l=1}^{2^b} \epsilon_l \ll 2^b C_0$ , which simplifies (B.3) to

$$\frac{V_{err}(y)}{V_{REF}} = \frac{\sum_{j=1}^y (2^b - y)\epsilon_j - \sum_{k=y+1}^{2^b} y\epsilon_k}{2^{2b} C_0}. \quad (B.4)$$

This error is the sum of independent zero-mean Gaussian random variables and thus is also Gaussian with a zero mean. The variance can be calculated by noting that all of the errors in the unit capacitors are independent with  $E[\epsilon_j^2] = \sigma_0^2$ .

$$\frac{E[V_{err}^2(y)]}{V_{REF}^2} = \frac{(2^b - y)^2 y + y^2 (2^b - y)}{2^{4b} C_0^2} \sigma_0^2 = \frac{y(2^b - y)}{2^{3b}} \cdot \frac{\sigma_0^2}{C_0^2} \quad (B.5)$$

The standard deviation of the INL, in terms of LSBs is

$$\sigma_{INL}(y) = \left( \frac{E[V_{err}^2(y)]}{V_{REF}^2} \right)^{1/2} \cdot 2^b = 2^{-b/2} \sqrt{y(2^b - y)} \frac{\sigma_0}{C_0} \text{ LSB}. \quad (B.6)$$

Figure B-1 shows that the behavioral model results from Fig. 4-11 match (B.6) for  $b = 5$  and  $\sigma_0/C_0 = 3\%$ .



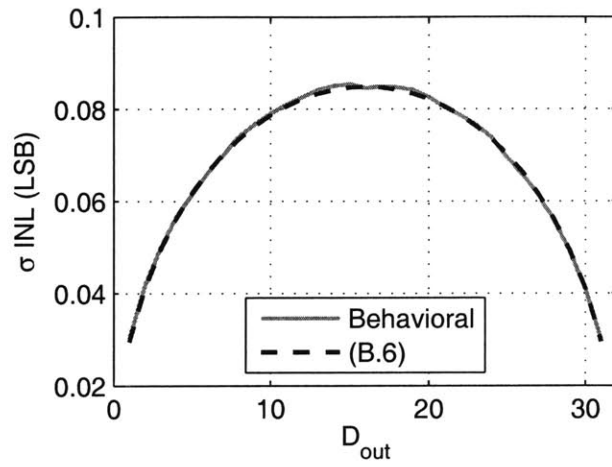


Figure B-1: Comparison of the behavioral model and the closed form expression derived for the INL of the SAR ADC with binary weighted or split capacitor arrays.



# Appendix C

## Energy Model Block Equations

Chapter 5 described the framework of the mixed-signal energy optimization, and it included highlights of the block energy models. Below are the details of these models, including the equations that determine the energy and output variables of the block. This appendix is very detailed and not for the faint of heart. For each block description, there is listed a table with the parameters required for that block (not including the global optimization parameters of Table 5.1). The parameter type is one of: **A** = global architectural parameters; **P** = process specific parameter; **I** = block input parameter; or, **O** = block output parameter. Any parameters listed in the equations that are not in the tables are local parameters (**L** in Fig. 5-3). After all the blocks are detailed, there is a summary of the required process data and set of simulations used to generate that data.

### C.1 Digital Logic

The core logic is simply the switching and leakage contributions from the SAR controller.

$$E_{dig,core} = \frac{b}{b_0} (C_{dig0} V_{DDD}^2 + I_{leak0} V_{DDD} t_{conv}) \quad (\text{C.1})$$

Here,  $C_{dig0}$  is an effective switched capacitance for the entire digital block and was taken from measurements of the first 65-nm testchip, with  $b_0 = 5$ . For ease of technology scaling,

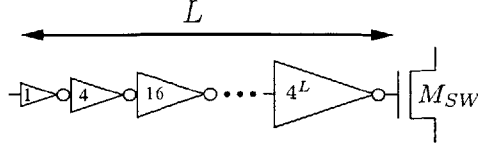


Figure C-1: Buffering chain used to drive the large capacitor array switches from the output of the control logic block. The total number of inverters inserted is  $L$ .

the digital energy can be referenced to a simple logic gate (e.g., an inverter with fan-out of 4 (FO-4)) times some architecture-dependent but process-independent multiplicative factor.  $I_{leak0}$  is the measured leakage current from the previous chip.

The second digital energy component is for driving the switches of the capacitor array, which are separated into the sampling switch and then bit-cycling switches. Buffering is automatically inserted to drive the large capacitive load of the switches as seen in Fig. C-1. The buffer chain assumes a scaling factor of 4 between successive inverters. Therefore, from the output of a minimum size inverter to the switch input, the number of stages that must be inserted is

$$L_k = \log_4 \frac{C_{sw,k}}{C_{FO4,1}} \quad (\text{C.2})$$

where  $k$  is *samp* or *bc* to represent the sampling or bit-cycling switch capacitive load. The capacitive load that can be driven by a single minimum size inverter is  $C_{FO4,1}$ . The number of buffers must be an integer, and a floor or ceiling function could be applied to (C.2), but the model is simpler if it is treated as a real number, and the error from not rounding is negligible. Then the total switched capacitance in the buffer chain, normalized to that of the minimum size inverter  $C_{sw,FO4}$ , is

$$\frac{C_{buf,k}}{C_{sw,FO4}} = \frac{1 - 4^{L_k+1}}{1 - 4}. \quad (\text{C.3})$$

Finally, the energy of the buffer chains is the sum of the switching energy and, for the sampling switch drive when  $V_{DDs} \neq V_{DD}$ , the energy of inserted level shifters, which is relatively small and could be neglected. Leakage power integrated over a conversion is also

included for all of these blocks, but it is omitted from the equations for the sake of brevity.

$$E_{buf,samp} = C_{buf,samp}V_{DDs}^2 + C_{sw,levelshift}V_{DDs}^2 \quad (C.4)$$

$$E_{buf,bc} = \frac{3}{4}C_{buf,bc}V_{DDs}^2 \quad (C.5)$$

The buffer energies and delays are referenced to that of a single minimum size FO-4 inverter. The coefficient 3/4 in (C.5) reflects the expected number of transitions for the bottom plate switches in the split capacitor array. The total energy is

$$E_{digital} = E_{dig,core} + E_{buf,samp} + E_{buf,bc}. \quad (C.6)$$

The charge pump supplies the current for the sampling buffer chain, which is

$$I_{load,cp} = \frac{E_{buf,samp}}{t_{conv}V_{DDs}}. \quad (C.7)$$

The setup and propagation delays through the digital logic are used to verify functionality and determine the amount of time available for the capacitor array and preamplifier settling.

$$t_{setup} = t_{s,DFP} + N_{setup}t_{d,FO4} \quad (C.8)$$

$$t_{pd,samp} = t_{c \rightarrow q}(V_{DDD}) + N_{samp}t_{d,FO4}(V_{DDD}) + t_{d,levelshift} + L_{samp}t_{d,FO4}(V_{DDs}) \quad (C.9)$$

$$t_{pd,bc} = t_{c \rightarrow q} + N_{bc}t_{d,FO4} + L_{bc}t_{d,FO4} \quad (C.10)$$

The terms  $N_{setup}$ ,  $N_{samp}$ , and  $N_{bc}$  represent the logic depths of the core digital logic that correspond to the setup time and sampling and bit decision propagation delays, respectively. These are architecture dependent and are 3, 4, and 1, respectively, for the logic controller described in Section 6.4.2.

The many parameters used in the digital logic block energy model are listed in Table C.1

Table C.1: Energy model parameters for the digital logic

PARAMETER	AIOP	DESCRIPTION
$b_0$	<b>A</b>	Resolution at which $C_{dig0}$ is measured/simulated
$C_{DIG0}$	<b>P</b>	Equivalent switched capacitance of the core digital logic, independent of the fan-out
$C_{FO4,1}$	<b>P</b>	Capacitive drive strength of a minimum size FO-4 inverter
$C_{sw,bc}$	<b>I</b>	Total capacitance of bottom-plate bit-cycling capacitors
$C_{sw,FO4}$	<b>P</b>	Switched capacitance of a minimum sized FO-4 inverter
$C_{sw,levelshift}$	<b>P</b>	Switched capacitance of the level shifter
$C_{sw,samp}$	<b>I</b>	Sampling switch input capacitance
$E_{digital}$	<b>O</b>	Total digital energy per conversion
$I_{load,CP}$	<b>O</b>	Sampling voltage current supplied by the charge pump
$I_{leak,FO4}(V_{DDD})$	<b>P</b>	Leakage current of a minimum size inverter
$I_{leak,levelshift}(V_{DDD})$	<b>P</b>	Leakage current of the level shifter
$I_{leak0}(V_{DDD})$	<b>P</b>	Leakage current of the core digital logic
$N_{bc}$	<b>A</b>	Logic depth between register output and bit-cycling buffer chain
$N_{samp}$	<b>A</b>	Logic depth between register output and sampling buffer chain
$N_{setup}$	<b>A</b>	Logic depth preceding register during bit decisions
$t_{c \rightarrow q}(V_{DDD})$	<b>P</b>	Clock edge to valid output delay of DFF
$t_{d,FO4}(V_{DDD})$	<b>P</b>	Propagation delay of FO-4 inverter
$t_{d,levelshift}(V_{DDD}, V_{DDS})$	<b>P</b>	Propagation delay of level shifter
$t_{pd,bc}$	<b>O</b>	Propagation delay from rising edge of clock to valid switch drives during bit decisions
$t_{pd,samp}$	<b>O</b>	Propagation delay from rising edge of clock to valid sampling switch drives
$t_{setup}$	<b>O</b>	Total setup time at end of bit decision periods
$t_{s,DFF}(V_{DDD})$	<b>P</b>	Setup time of DFF

## C.2 Capacitor Array

The nominal array capacitance,  $C_T$ , is determined from the matching constraints in (2.3) with  $\zeta = 0.5$ .

$$C_T = 2^{2^b - b'_0} C'_0 \quad (\text{C.11})$$

The capacitance  $C'_0$  is the unit capacitance required for matching to the  $b'_0$  level. This total nominal capacitor array size is used to determine the sampling and bit-cycling bottom plate

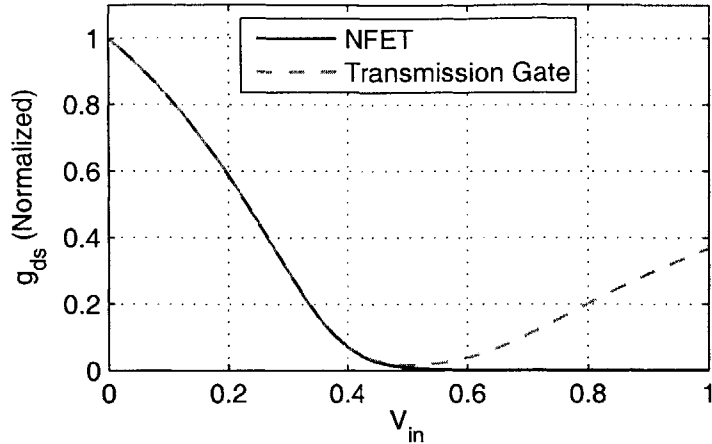


Figure C-2: Normalized conductance,  $g_{ds}$ , of the NFET and transmission gate bottom plate switch versus the input voltage.

switch sizes.

### C.2.1 Sampling

The conductances of the NFET and TG sampling networks are plotted in Fig. C-2 versus the input voltage. As shown in the figure, NFET sampling can only be used if the reference voltage is confined to be low. The step response portion of sampling requires that

$$t_{sample} > k_{slew}(b + 1) \ln(2)\tau_{samp}. \quad (C.12)$$

The factor  $k_{slew}$  is used to model the nonlinear slewing portion at the onset of sampling and is determined from simulations to be 1.9. The time constant for sampling is approximately  $\tau_{samp} = C_S/g_{ds,min}$ , where  $g_{ds,min}$  is the minimum conductance of the switch (from Fig. C-2) between 0 and  $V_{REF}$ .

Accurately tracking a high frequency input signal is more complicated because the non-linearity arises only from the variation of the conductance across the input range. A flat conductance, independent of its value, will not produce any harmonic distortion. Volterra analysis [138] could be used to formulate a closed form expression for the distortion, which

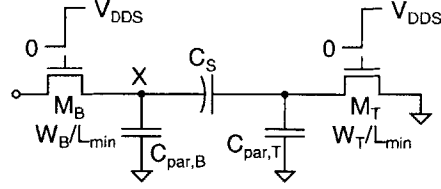


Figure C-3: NFET sampling network with associated parasitic capacitances  $C_{par,B} \propto W_B$  and  $C_{par,T} \propto W_T$ .

can then lead to proper sizing of the switches; however, this approach is highly dependent upon the particular device characteristics and makes the model much more difficult to setup and transition to new process nodes. Instead, this work uses a much simpler model that produces a conservatively sized sampling switch. The most attenuation of an input signal across  $S_B$  occurs at the maximum frequency and at the voltage where the switch conductance is minimum. By restraining the attenuation at this point to less than half an LSB, the attenuation across the entire input signal range will be no more than this, and the input will be accurately tracked across the bottom plate switch. This condition is quantified in (C.13) and has been verified by simulations to produce a conservatively sized sampling network. The simulations predict that the network has a total harmonic distortion (THD)  $< -43$  dBc across the entire range of  $V_{DD}$  and  $V_{REF}$  values, sufficient for a 5-bit ADC.

$$\frac{1}{\sqrt{1 + (f_S \tau_{samp}/2)^2}} > 1 - 2^{-(b+1)} \quad (\text{C.13})$$

The stricter bound of (C.12) and (C.13) sets the maximum  $\tau_{samp}$ . The sizing of the NFET sampling network to meet this bound is discussed here. The optimization of the transmission gate network follows a similar process but is more complex because of an added free variable; it is discussed in Appendix D. For a fixed  $V_{REF}$ ,  $V_{DD}$ , and  $\tau_{max}$ , the two remaining free variables are the widths of the bottom and top plate switches,  $W_B$  and  $W_T$ , respectively. The effect of the parasitic capacitance of the switches is added, as shown in Fig. C-3, where  $C_{par,k} \propto W_k$  for  $k \in \{B, T\}$ . For the bandwidth constraint (C.13), the total time constant



of the sampling network is determined using open circuit time constants (OCTC) as

$$\tau_{samp} = C_S \left( \frac{1}{g_{ds,T}} + \frac{1}{g_{ds,B}} \right) + \frac{C_{par,T}}{g_{ds,T}} + \frac{C_{par,B}}{g_{ds,B}}. \quad (\text{C.14})$$

The bottom plate switch conductance is  $g_{ds,B} = \min_{V_S} g_{ds}(V_G, V_S) = g_{ds}(V_{DD}, V_{REF})$  because the minimum conductance always occurs at the largest input signal for the NFET switch. The top plate switch conductance is simply  $g_{ds,T} = g_{ds}(V_{DD}, 0)$ . The last two terms in (C.14) are independent of the switch widths and are only dependent upon the relevant voltages in the circuit. Thus, to determine  $W_T$  and  $W_B$ , an effective time constant

$$\tau_{eff} = \tau_{samp} - \frac{C_{par0,T}}{g_{ds0,T}} - \frac{C_{par0,B}}{g_{ds0,B}} \quad (\text{C.15})$$

is used. The sampling network model fails if  $\tau_{eff} < 0$ . It should be noted that OCTC is only useful to find a conservative lower bound on the 3 dB bandwidth and is not directly related to settling time of a non-first-order system; however, for most designs, including the optimal one,  $C_S \gg C_{par,T} + C_{par,B}$  and the sampling network is effectively first order, making the error from using  $\tau_{eff}$  in both the tracking and settling time constraints negligible.<sup>1</sup> Using (C.14), (C.15), and  $g_{ds,k} = (W/W_0)g_{ds0,k}$ ,  $W_B$  can be expressed in terms of  $W_T$  as

$$W_B = \frac{1}{\left( \frac{\tau_{eff}}{C_S W_0} - \frac{1}{W_T g_{ds0,T}} \right) g_{ds0,B}}. \quad (\text{C.16})$$

The energy driving the sampling switch is proportional to the total width of the sampling switch, which is  $W_B + 4W_T$ . The factor 4 is specific to the hierarchical top plate sampling network described in Section 6.3.1. The energy is minimized when

$$W_T = \frac{C_S W_0}{\tau_{eff}} \left[ \frac{1}{g_{ds0,T}} + (4g_{ds0,B}g_{ds0,T})^{-1/2} \right], \quad (\text{C.17})$$

and  $W_B$  is calculated from (C.16).

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<sup>1</sup>For the optimization without a charge pump (top curve in Fig. 5-12), the capacitive self-loading of the switches becomes significant, making this approximation less accurate.

Finally, the parasitic capacitance seen by  $V_{in}$  when the array is not sampling is

$$C_{off} = \frac{W_B}{W_0} C_{off0}, \quad (\text{C.18})$$

which is used for the input buffer model (Section C.8).

## C.2.2 Bit Cycling

The bottom plate switches connected to  $V_{REF}$  and ground are sized such that the capacitor array settles within the available time at the beginning of every bit decision period. The settling time requirement is

$$t_{settle,bc} = (k_{slew}(b+1) \ln 2) \tau_{bc}. \quad (\text{C.19})$$

The time constant  $\tau_{bc} = C_T/g_{ds}$  depends on the specific switch configuration and voltages used for the bottom plate switches and is modified to include the self-loading of the switch, as in (C.15). The switch to ground is always a simple NFET with  $g_{ds,gnd} = g_{ds0}(V_G = V_{DDD}, V_S = 0)W_{gnd}/W_0$ . The switch connected to the reference voltage can be either an NFET or PFET with conductance  $g_{ds,ref} = g_{ds0}(V_G = V_{DDD}, V_S = V_{REF}, V_B = 0)W_{ref}/W_0$  or  $g_{ds,ref} = g_{ds0p}(V_G = 0, V_S = V_B = V_{REF})W_{ref}/W_0$  respectively. Depending on the appropriate supply and reference voltages, the switch with the larger  $g_{ds}$  per unit width is chosen. These widths are the total widths of all the bottom plate switches. This analysis assumes the total switch width is divided among the capacitors in the array inversely proportional to the capacitor size (i.e., the switch connected to the MSB capacitor is twice the size of the MSB-1 capacitor).

The total switch load is calculated in (C.20), which is used by the digital logic to appropriately buffer the switch signals.

$$\begin{aligned} C_{sw,samp} &= \frac{W_B + 4W_T}{W_0} C_{gs0} \\ C_{sw,bc} &= \frac{W_{gnd} + W_{ref}}{W_0} C_{gs0} \end{aligned} \quad (\text{C.20})$$

Table C.2: Energy model parameters for the capacitor array

PARAMETER	AIOP	DESCRIPTION
$b'_0$	<b>P</b>	Resolution at which $C'_0$ is determined
$C'_0$	<b>P</b>	Capacitance required for matching to the $b'_0$ level
$C_{gs0}$	<b>P</b>	Equivalent gate capacitance of unit FET switch
$C_{off}$	<b>O</b>	Parasitic capacitance connected to $V_{in}$ when sampling switches are open
$C_{off0}$	<b>P</b>	Parasitic capacitance of the unit NFET switch while off
$C_{par0}$	<b>P</b>	Parasitic capacitance of the unit NFET switch while turned on
$C_{sw,samp}$	<b>O</b>	Sampling switch load capacitance
$C_{sw,bc}$	<b>O</b>	Total bit-cycling bottom plate switch load capacitance
$E_{caparray}$	<b>O</b>	Switching energy of the array from the $V_{REF}$ supply
$g_{ds0}(V_G, V_S, V_B)$	<b>P</b>	On conductance of unit NFET switch
$g_{ds0p}(V_G, V_S, V_B)$	<b>P</b>	On conductance of unit PFET switch
$k_{stew}$	<b>P</b>	Fitting parameter for nonlinear settling
$t_{settle,bc}$	<b>I</b>	Time available for capacitor array settling during bit decision periods
$t_{sample}$	<b>I</b>	Time available for sampling
$W_0$	<b>P</b>	Width of the unit FET used for rest of process data

The total energy drawn from  $V_{REF}$  required to switch the capacitor array is

$$E_{caparray} = 2\eta(C_T + C_{par,B} + C_{par,T} + C_{par,gnd} + C_{par,ref})V_{DDA}V_{ref}, \quad (\text{C.21})$$

similar to (2.2) with the addition of the switch parasitic capacitances.

The parameters for the capacitor array energy model described here are listed in Table C.2. The additional parameters required for the transmission gate bottom plate sampling switch are listed separately within Appendix D.

## C.3 Comparator

### C.3.1 Preamplifier

The preamplifier model in Chapter 2 (Fig. 2-5) is expanded here to include the effects of parasitic wiring capacitances, optimization of the bias settings, and the self-loading of the input differential pair's drain capacitance and output resistance. The inputs to the preamplifier block are the gain, settling time, and load capacitance (from the following preamplifier or regenerative latch). The effective load capacitance is modified to include a fixed wiring capacitance,  $C'_L = C_L + C_{wire}$ . The total gain of the preamplifier is

$$A_V = g_m (R_L \parallel r_o), \quad (\text{C.22})$$

where  $r_o$  is the output resistance of  $M_1$ – $M_2$ . The speed is

$$\tau = (R_L \parallel r_o) \cdot (C'_L + C_{db1} + C_{gd1}). \quad (\text{C.23})$$

In addition to the load capacitance, there is the self-loading effect from the parasitic capacitance at the drain of  $M_1$ . For a given current density, the device parameters scale with the width as  $g_m, C_{gs}, C_{db}, C_{gd} \propto W$  and  $r_o \propto 1/W$ . Thus, if all of the process specific device parameters are extracted with a nominal device width of  $W_0$ , the required device width to meet the settling time and gain specifications is

$$W = \frac{C'_L W_0}{\frac{\tau}{A_V} g_{m0} - C_{d0}}. \quad (\text{C.24})$$

The total drain capacitance is  $C_d = C_{db} + C_{gd}$ . The “0” index above indicates the parameter value at width  $W_0$ .

Finally, the load resistance, input capacitance, bias current, and power can be calculated

as

$$R_L = \frac{A_V r_{o0}}{W/W_0(g_{m0}r_{o0} - A_V)}, \quad (\text{C.25})$$

$$C_{in} = \frac{W}{W_0} C_{in0}, \quad (\text{C.26})$$

$$I_{bias} = 2k_{bias} \frac{W}{W_0} I_{D0}, \quad \text{and} \quad (\text{C.27})$$

$$E_{preamp} = I_{bias} V_{DD} A t_{conv}. \quad (\text{C.28})$$

Because the gain of the preamplifier is low, the Miller effect can be neglected, and  $C_{in0} \approx C_{gs0}$ . The scaling factor  $k_{bias}$  is used to model any extra current cost due to a biasing network and is typically between 1 and 2.

First, it should be noted that not all input specifications are achievable with this preamplifier topology. If the denominators of (C.24) or (C.25) go to zero or negative, the preamplifier model fails. The former indicates that the gain bandwidth product is too high. Note that unlike in the model in Section 2.1.3, the high frequency limit is based on  $g_m/C_d \gg f_T \approx g_m/C_{gs}$ . In practice, however, the chosen operating point is still well below either of these frequencies. The second failure condition relates to the the maximum intrinsic gain of a transistor. If length were added as a separate free variable, the intrinsic gain could be increased at the cost of increased capacitance and power.

Finally, the bias point of the preamplifier can be chosen to minimize  $I_{bias}$  subject to the failure constraint. All of the transistor-dependent parameters ( $g_{m0}$ ,  $C_{gs0}$ ,  $C_{d0}$ ,  $r_{o0}$ ) are extracted for different gate, body, and drain voltages. Three bias optimization scenarios are supported in this model. The first is that the input common mode voltage,  $V_{in,cm}$ , and common source voltage,  $V_{CS}$ , are fixed, as would be necessitated if the input common mode level is fixed by an available reference voltage, and the preamplifier is configured in a grounded-source pseudo-differential configuration. The second bias configuration fixes the source voltage (e.g., pseudo-differential) but allows the input common mode voltage to vary. The final bias option varies both the input common mode and common source voltage, but leaving a minimum amount of headroom for a tail current source transistor,  $V_{CS} > V_{CS,min}$ .

Table C.3: Energy model parameters for the preamplifier

PARAMETER	AIOP	DESCRIPTION
$A_V$	<b>I</b>	Required preamplifier gain
$C_{d0}(V_G, V_D)$	<b>P</b>	Unit drain capacitance, $C_{gd0} + C_{db0}$
$C_{in}$	<b>O</b>	Input capacitance (load for previous stage)
$C_{in0}(V_G, V_S)$	<b>P</b>	Unit input capacitance ( $\approx C_{gs0}$ )
$C_L$	<b>I</b>	Load capacitance from following preamplifier or latch
$C_{wire}$	<b>P</b>	Fixed parasitic wiring capacitance
$E_{preamp}$	<b>O</b>	Preamplifier energy per conversion
$g_{m0}(V_G, V_S)$	<b>P</b>	Unit transconductance
$I_{D0}(V_G, V_S)$	<b>P</b>	Drain current of unit transistor
$k_{bias}$	<b>A</b>	Current scaling factor for bias network overhead
$r_{o0}(V_G, V_S)$	<b>P</b>	Unit transistor small signal output resistance
$\tau$	<b>I</b>	Required preamplifier time constant
$V_{CS}$	<b>A</b>	Fixed or minimum common source voltage of the input differential pair
$V_{CM,in}$	<b>A/O</b>	Input common mode voltage, either fixed or selected during bias optimization
$W_0$	<b>P</b>	Unit width at which device parameters were extracted

When possible, particularly at large levels of parallelism, the optimization moves the bias point from strong inversion to moderate inversion for improved transconductance and energy efficiencies [139].

The parameters for the preamplifier model are listed in Table C.3.

### C.3.2 Latch

The latch model implements the matching model from [70] wherein matching can be traded off directly with device size, with  $\sigma_{OS} \propto 1/\sqrt{WL}$ . The input capacitance and switching energy are directly proportional to the device size.

$$E_{latch} = \frac{W}{W_0} b C_{sw0} V_{DDD}^2 \quad (\text{C.29})$$

$$C_{in} = \frac{W}{W_0} C_{in0} \quad (\text{C.30})$$

$$V_{OS} = \left( \frac{W}{W_0} \right)^{-1/2} V_{OS0} \quad (\text{C.31})$$

Table C.4: Energy model parameters for the latch

PARAMETER	AIOP	DESCRIPTION
$C_{in}$	<b>O</b>	Latch input capacitance
$C_{in0}$	<b>P</b>	Input capacitance for width $W_0$
$C_{sw0}(V_{DDD})$	<b>P</b>	Switched capacitance for width $W_0$
$E_{latch}$	<b>O</b>	Total latch energy per conversion
$t_d$	<b>O</b>	Latch delay
$t_{d0}(V_{DDD})$	<b>P</b>	Total regeneration time of the latch
$V_{OS0}(V_{DDD})$	<b>P</b>	Input offset voltage for width $W_0$
$W$	<b>I</b>	Width, input from the comparator block
$W_0$	<b>P</b>	Nominal width for process data

The latch also has an unbounded delay associated with it, dependent on both the supply voltage and the differential input voltage (see (4.27)); however, a worst case delay for a given bit error rate is obtained from simulations ( $t_d = t_{d0}(V_{DDD})$ ).

Table C.4 lists the parameters for the latch model. With the exception of  $W_0$ , all of the process parameters have some voltage supply dependence. Of these, only  $t_{d0}$  has a strong supply dependence; the remainder could be modeled as voltage independent with only a small degradation in the accuracy of the model. Further, to facilitate scaling across process technologies, the parameters  $t_{d0}$ ,  $I_{leak0}$ , and  $C_{sw0}$  can all be referenced to that of a standard digital gate (see Section 5.3.1) with process-independent scaling factors.

### C.3.3 Comparator Equations

The entire comparator chain is formed from a cascade of preamplifiers, each with its own OOS, followed by a regenerative latch. At the comparator level, an optimization is performed to choose the number of preamplifiers and latch width (or equivalently its offset voltage). For a given latch offset voltage,  $V_{OS,latch}$ , the preamplifiers are solved working backwards, from the last preamplifier to the first. The load capacitance of preamplifier  $i$  is  $C_{L,i} = (1 + k_{ccb})\nu C_{in,i+1}$ , which includes the capacitor divider from the autozeroing capacitor  $C_C$  and input capacitance of the following stage.  $\nu = C_{Ci}/(C_{Ci} + C_{in,i+1})$  is the same as defined in Section 2.1.3. The parasitic bottom plate capacitance of  $C_C$ , which scales with both  $C_C$

(and therefore  $C_{in,i+1}$  for a fixed  $\nu$ ) is modeled with  $k_{ccbp}$ . Each preamplifier is assigned the same gain and time constant, although the individual sizes and bias conditions may be different due to the changing load capacitance down the chain. The required gain and time constant of each preamplifier in the chain are

$$A_V = \frac{1}{\nu} \left( \frac{V_{OS,latch}}{V_{OS,in}} \right)^{1/N_{preamp}} \quad (\text{C.32})$$

$$\tau = \frac{t_{settle}}{k_{slew}(b+1)\ln(2)N_{preamp}}. \quad (\text{C.33})$$

The required comparator input offset voltage,  $V_{OSin}$ , is set relative to the LSB voltage as discussed in Section 5.3.4. The settling time is divided evenly among the preamplifiers. In (C.33), the term  $k_{slew}$  models any nonlinear settling behavior; its value is set to be 1.9, as determined with simulations.

The total comparator energy is

$$E_{comparator} = E_{latch}(V_{OS,latch}) + \sum_{i=1}^{N_{preamp}} E_{preamp,i}. \quad (\text{C.34})$$

The outputs of the comparator model are shown in Fig. C-4. While a relatively simple model, it still captures the expected behavior that it is more energy efficient to get gain from cascaded stages as opposed to a single high gain stage. The model limits the gain per stage to within 3 to 10. Table C.5 lists the comparator model parameters.

## C.4 Digital Offset Correction

The energy cost of the digital offset correction is added as

$$E_{offset} = bC_{sw,fulladd}V_{DD}^2. \quad (\text{C.35})$$

For increased process independence, the switched capacitance of the full adder,  $C_{sw,fulladd}$ , can be referenced to that of the standard inverter,  $C_{SW,FO4}$ . In addition, if the offset is being



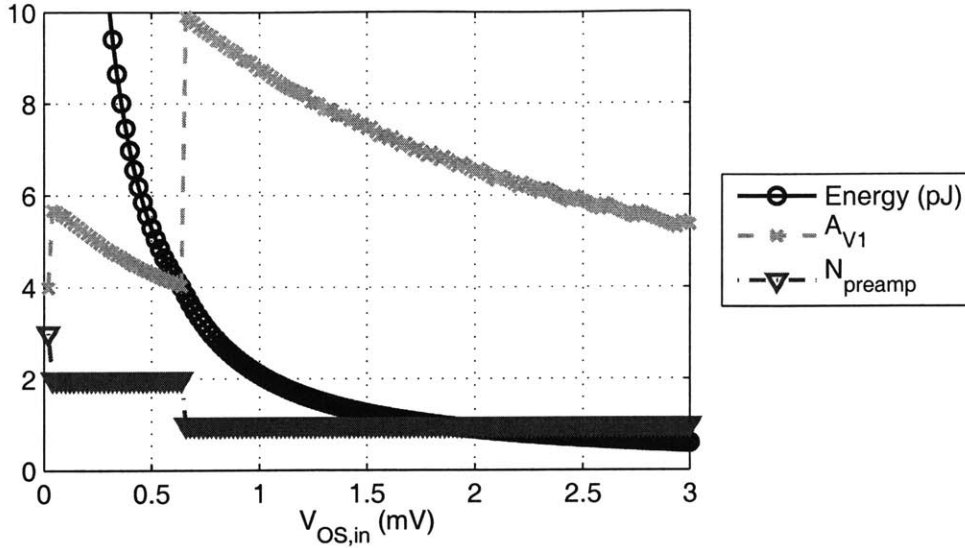


Figure C-4: Outputs of the comparator model as the required input referred offset voltage is varied. Shown are the number of preamplifiers  $N_{preamp}$ , gain per preamplifier  $A_{V1}$ , and total comparator energy  $E_{comp}$ .

Table C.5: Energy model parameters for the comparator

PARAMETER	AIOP	DESCRIPTION
$A_V$	<b>O</b>	Gain of a single preamplifier
$E_{comparator}$	<b>O</b>	Comparator energy per conversion
$k_{ccbp}$	<b>P</b>	Coupling capacitor parasitic bottom plate capacitance ratio
$k_{slew}$	<b>P</b>	Nonlinear settling scaling coefficient for preamplifiers
$\nu$	<b>A</b>	Voltage loss across autozeroing capacitor, chosen to be 5/6 for implementation purposes
$t_{settle}$	<b>I</b>	Settling time available for the preamplifiers
$V_{OS,in}$	<b>I</b>	Required comparator input offset voltage

corrected to a sub-LSB level, the width of the output mux needs to be increased (i.e., for correcting to an LSB/2,  $b$  in (5.9) increases to  $b + 1$ ).

Table C.6 lists the parameters for the offset correction model.

Table C.6: Energy model parameters for the digital offset correction

PARAMETER	AIOP	DESCRIPTION
$C_{sw,fulladd}$	<b>P</b>	Switched capacitance of a full adder
$E_{offset}$	<b>O</b>	Digital offset correction energy per conversion

Table C.7: Energy model parameters for the channel

PARAMETER	AIOP	DESCRIPTION
$\delta_{t_{cap}}$	<b>A</b>	Fractional portion of period dedicated to capacitor array settling
$\delta_{t_{samp}}$	<b>A</b>	Fraction of sampling period used for active sampling
$t_{sample}$	<b>O</b>	Window for settling of the capacitor array during sampling
$t_{settle,bc}$	<b>O</b>	Capacitor array settling time during bit decisions
$t_{settle,pre}$	<b>O</b>	Available preamplifier settling time

## C.5 Channel

The channel block itself does not consume any energy, and for modeling purposes, it simply enforces the timing of the channel sub-blocks, as described in Section 5.3.5.

When setting up the timing constraints (i.e.,  $t_{settle,bc} = \delta_{t_{cap}}(t_{per}/2 - t_{pd,bc})$ ), there is a positive feedback loop between the digital logic and the capacitor array. The propagation delay through the logic grows with the size of the capacitor array switches ( $L_k$  in (C.9)–(C.10) increases with  $C_{sw,k}$ ; see (C.2)), but this reduces the amount of time available for the settling of the capacitor array. With less settling time, the switch time constant decreases (see (C.12) and (C.19)), and its width must increase, further increasing the load capacitance seen by the buffers at the output of the digital block. This feedback loop may not be closed for the sampling operation if (C.13) is a stricter bound than (C.12), but it is always present for the bit decision timing. The digital logic and capacitor array blocks are solved iteratively until the total timing converges, as shown in Fig. C-5. If this loop does not converge, then the model fails for that input condition. In addition, the model fails if  $t_{d,latch} + t_{setup} > t_{per}/2$ , when there is not enough time in the second half of the clock period to meet the latch delay and setup time constraints.

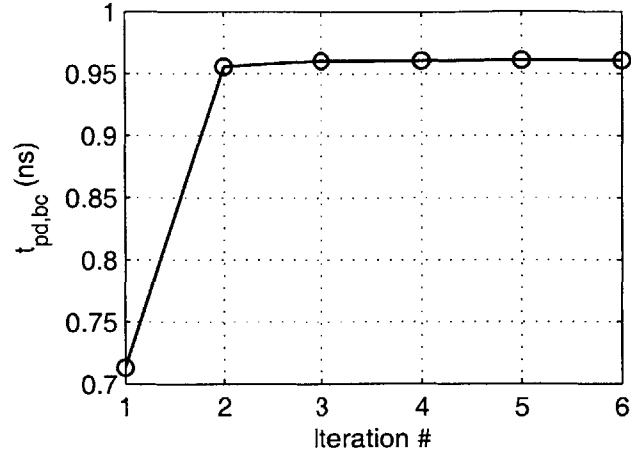


Figure C-5: The modeled digital propagation delay as the channel block iterates the logic and capacitor array block solutions. The timing converges at iteration 4.

Table C.8: Energy model parameters for the clock distribution

PARAMETER	AIOP	DESCRIPTION
$C_{clk0}$	<b>P</b>	Clock wiring capacitance to a single channel
$E_{clk}$	<b>O</b>	Energy per conversion of the clock distribution
$M_0$	<b>P</b>	# of channels where $C_{clk0}$ was determined

The timing parameters local to the channel model are listed in Table C.7.

## C.6 Clock Distribution

The clock’s energy model equations are presented in Section 5.3.6. Table C.8 lists the clock distribution parameters.

## C.7 Output Mux

The output mux parameters are listed in Table C.9

Table C.9: Energy model parameters for the output mux

PARAMETER	AIOP	DESCRIPTION
$C_{sw,mux}$	<b>P</b>	Switched and wiring capacitance of a single mux stage
$E_{mux}$	<b>O</b>	Energy per conversion of the output mux
$k_{mux}$	<b>A</b>	Muxing factor of each stage in the output mux tree

## C.8 Input Buffer

The total input capacitance is

$$C_{in} = M \cdot \left[ C'_T \frac{t_{sample}}{t_{conv}} + C_{off} \left( 1 - \frac{t_{sample}}{t_{conv}} \right) + C_{wire,in} \right], \quad (\text{C.36})$$

which is the sum of the total array capacitance and bottom plate capacitance ( $C'_T$ ) for the channels actively sampling plus the off capacitance of all channels that are not actively sampling. In (C.36), the average input capacitance of a single channel over a conversion is first calculated and then multiplied by the total number of channels. The routing capacitance to a single channel is denoted as  $C_{wire,in}$ . To minimize skew between channels, the routing network for  $V_{in}$  must be balanced to every channel, similar to the clock distribution requirements (5.6), and  $C_{wire,in} = C_{wire,in0}(M/M_0)^{1/2}$

There are many possible topologies that can serve as input buffers. The source follower is a simple open loop topology with an output impedance  $1/g_m$  that must be set such that  $g_m/C_{in} \geq 2\pi f_S/2$ .

The input buffer energy is

$$E_{inbuf} = \begin{cases} 0 & C_{in} \text{ small} \\ k_{buf} \frac{I_{D0}}{g_{m0}} C_{in} 2\pi f_S / 2 k_{backoff} & \text{o.w.}, \end{cases} \quad (\text{C.37})$$

where  $k_{buf}$  is a topology-specific constant, which can be 1 for the source follower. All of the transistor specific parameters, ( $g_{m0}$ ,  $I_{D0}$ , etc.) are identical to those used in the preamplifier model. The model is further extended to include the parasitic capacitance of the buffer itself. If an input buffer is necessary, the power of the input buffer is actually larger than the rest

Table C.10: Energy model parameters for the input buffer

PARAMETER	AIOP	DESCRIPTION
$C_{wire,in0}$	<b>P</b>	Routing capacitance of input network to a channel
$E_{inbuf}$	<b>O</b>	Input buffer energy per conversion
$k_{backoff}$	<b>A</b>	Amount of backoff between input buffer speed and Nyquist input frequency
$k_{buf}$	<b>A</b>	Input buffer topology-specific current scaling factor
$M_0$	<b>P</b>	Number of channels at which $C_{wire,in0}$ is determined
$R_S$	<b>A</b>	Source resistance of circuit driving the ADC
$t_{sample}$	<b>I</b>	Total sampling time

Table C.11: Energy model parameters for the charge pump

PARAMETER	AIOP	DESCRIPTION
$E_{CP}$	<b>O</b>	Charge pump energy per conversion
$\eta_{CP}$	<b>A</b>	Charge pump efficiency
$I_{loadcp}$	<b>I</b>	Charge pump load current per channel

of the ADC, making the condition (5.10), combined with (C.36), a hard upper bound on  $M$ .

Table C.10 lists the parameters for the input buffer model.

## C.9 Charge Pump

The charge pump energy is

$$E_{CP} = \left( \frac{1}{\eta_{CP}} - 1 \right) \frac{MI_{loadcp}V_{DDs}}{f_s}. \tag{C.38}$$

The charge pump model parameters are listed in Table C.11.

## C.10 Summary of Process Parameters

The energy model requires 35 process parameters, marked **P** in the tables above. These are summarized in Table C.12, along with the simulation that can be used to generate these parameters. While the number of parameters is large, most can be obtained from

Table C.12: Process parameters required for energy model

Parameter	Block	Simulation	Parameter	Block	Simulation
$b'_0$	Array	Matching	$g_{ds0}$	Array	NFET
$C'_0$	Array	Matching	$g_{ds0p}$	Array	PFET
$C_{clk0}$	Clock	Parasitics	$g_{m0}$	Preamp	NFET
$C_{d0}$	Preamp	NFET	$I_{D0}$	Preamp	NFET
$C_{DIG0}$	Digital	See below	$I_{leak0}$	Latch	Latch
$C_{FO4,1}$	Digital	Inverter	$I_{leak,FO4}$	Digital	Inverter
$C_{gs0}$	Array	NFET	$I_{leak0}$	Digital	See below
$C_{in0}$	Latch	Latch	$k_{ccb}$	Comparator	Parasitics
$C_{in0}$	Preamp	NFET	$k_{slew}$	Array/Preamp	See below
$C_{off0}$	Array	NFET	$M_0$	Clock	Parasitics
$C_{par0}$	Array	NFET	$r_{o0}$	Preamp	NFET
$C_{sw0}$	Array	NFET	$t_{c \rightarrow q}$	Digital	DFF
$C_{sw,FO4}$	Digital	Inverter	$t_{d0}$	Latch	Latch
$C_{sw,fulladd}$	Offset	Adder	$t_{d,FO4}$	Digital	Inverter
$C_{sw,levelshift}$	Digital	Level shift	$t_{d,levelshift}$	Digital	Level shift
$C_{sw,mux}$	Mux	Mux/Par.	$V_{OS0}$	Latch	Matching
$C_{wire}$	Preamp	Parasitics	$W_0$		NFET
$C_{wire,in0}$	Buffer	Parasitics			

simulations or measurements of a single NFET or PFET, or a simple digital gate (inverter, DFF, adder, level shifter, mux, latch). Process variation data is necessary to properly size the capacitor array and the latch. The technology vendor provided transistor variation data, and the capacitor matching was extracted from measurements of the earlier 65-nm testchip. Several parameters require some knowledge of wiring parasitics, which for this work were taken from extracted parasitics and measurements from the first 65-nm testchip. Parasitics critically affect both the energy and speed of circuits in deep sub-micron CMOS. Even local wiring parasitics within a digital gate can have a significant effect. All of the capacitance and delay parameters were either derived from simulations with extracted parasitics, or were multiplied by a constant scaling factor that was determined by comparing simulations of the basic preamplifier and inverter with and without extracted parasitics.

Finally, a few parameters merit special consideration. All of the step response transient equations, (C.12), and (C.19), and (C.33), use the constant  $k_{slew}$  to model any nonlinear

or slewing effects that lengthen the settling time. This can be separately determined by simulations of the relevant blocks (i.e., capacitor array switches and preamplifier); however, all of these simulations indicate that a value between 1.5 and 2 is appropriate. The exact number will vary somewhat depending on device nonlinearities and the specific voltages of the transient, but choosing a value within this range is appropriate and should hold reasonably well across technology generations. The core digital switching energy and leakage current density,  $C_{DIG0}$  and  $I_{leak0}$ , are the only parameters that require simulation or measurement of a complex block. For the implemented model, this was not a limitation because a complete logic controller had already been implemented in the same technology for the earlier testchip; however, these parameters can be referenced to those of a simple inverter, reducing the complexity of the model setup and simulation for a new technology. Specifically, for the implemented model  $C_{DIG0} = 1100C_{sw,FO4}$ , and  $I_{leak0} = 655I_{leak,FO4}$ .





# Appendix D

## Transmission Gate Sampling Energy Model

Appendix C.2.1 analyzed the sampling network with a boosted NFET bottom plate switch (Fig. C-3). The transmission gate (Fig. 5-5(b)) is an alternative bottom plate switch implementation that can be used with larger full scale input voltages. Unfortunately, the addition of the PFET introduces an extra free variable in the optimization, as there are three widths to jointly optimize to meet the sampling time constraint. This appendix presents the optimization of the transmission gate sampling network, including the energy model setup and the techniques that were used for solving this model in Matlab.

Referring to Fig. D-1, the parasitic capacitances, conductances, and switch self-sampling time constants are specified in (D.1)–(D.6). The effective time constant of the top plate switch,  $\tau_T$ , is dependent only on the voltage settings, but the bottom plate time constant,

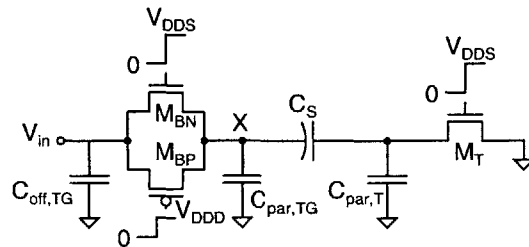


Figure D-1: Transmission gate sampling network with parasitics.

$\tau_{TG}$ , depends on the widths of the two transistors in the transmission gate.

$$C_{par,T} = \frac{W_T}{W_0} C_{par0n} \quad (D.1)$$

$$C_{par,TG} = C_{par,BN} + C_{par,BP} = \frac{W_{BN}}{W_0} C_{par0n} + \frac{W_{BP}}{W_0} C_{par0p} \quad (D.2)$$

$$g_{ds,T} = \frac{W_T}{W_0} g_{ds0n}(V_{DD}, 0) \quad (D.3)$$

$$g_{ds,TG} = g_{ds,BN} + g_{ds,BP} = \frac{W_{bn}g_{ds0n}(V_{DD}, V_{in}) + W_{BP}g_{ds0p}(0, V_{in})}{W_0} \quad (D.4)$$

$$\tau_T = \frac{C_{par,T}}{g_{ds,T}} = \frac{C_{par0n}}{g_{ds0n}(V_{DD}, 0)} \quad (D.5)$$

$$\tau_{TG} = \frac{C_{par,TG}}{g_{ds,TG}} = \frac{C_{par,BN} + C_{par,BP}}{g_{ds,BN} + g_{ds,BP}} \quad (D.6)$$

Looking more closely at (D.6), the bottom plate time constant is dependent only on the ratio of the widths. Defining  $x$  as the ratio of the PFET width to the NFET width,  $x = W_{bp}/W_{bn}$ , the transmission gate inherent sampling time constant is

$$\tau_{TG}(V_{in}) = \frac{C_{par0n}W_{bn} + C_{par0p}W_{bp}}{g_{ds0n}W_{bn} + g_{ds0p}W_{bp}} = \frac{C_{par0n} + xC_{par0p}}{g_{ds0n} + xg_{ds0p}}. \quad (D.7)$$

This time constant depends on  $V_{in}$  because  $g_{ds0}$  is voltage dependent (Fig. C-2). Consistent with the analysis presented in Appendix C.2.1, the relevant conductance and time constant used in computing the widths of the devices is

$$g_{ds0,TG} = \min_{0 < V_{in} \leq V_{REF}} g_{ds0n} + xg_{ds0p} \quad (D.8)$$

$$\tau_{TG} = \max_{0 < V_{in} \leq V_{REF}} \tau_{TG}(V_{in}).$$

Both of these limits occur at roughly the same value of  $V_{in}$  because  $C_{par0}$  is, to first order, independent of the relevant voltages.

The effective time constant (C.15) is

$$\tau_{eff,TG} = \tau_{samp} - \tau_{TG} - \tau_T \quad (D.9)$$

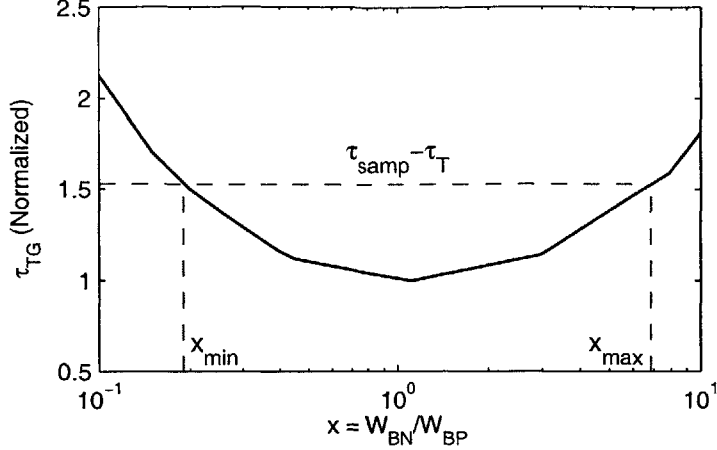


Figure D-2: Transmission gate sampling time constant as a function of the N/PFET ratio. The horizontal dashed line corresponds to (D.9) going to 0.

for the transmission gate switch. Figure D-2 plots the transmission gate switch time constant versus  $x$ , along with  $\tau_{samp} - \tau_T$ , the maximum allowable time constant before this switch fails. The minimum and maximum values of  $x$  that satisfy this limit are  $x_{min}$  and  $x_{max}$ , respectively.

Given the widths of any two of the switches, the third can be calculated to meet the settling time constraint. Specifically, if  $W_{BP}$  and  $W_{BN}$  or, equivalently,  $W_{BN}$  and  $x$  are specified, then  $W_T$  can be calculated as

$$W_T = \frac{W_0}{g_{ds0n}(V_{DDs}, 0) \left[ \frac{\tau_{eff}}{C_T} - \frac{1}{W_{BN}g_{ds0,TG}} \right]}. \quad (\text{D.10})$$

The cost function (D.11) for the transmission gate switch is more complicated than the NFET case because the top and bottom plate NFETs are driven by  $V_{DDs}$  to maximize their overdrive voltage, while the PFET is driven by  $V_{DDd}$  to minimize its energy, and the unit gate capacitance may be different for the two switches. All three widths can be determined by minimizing (D.11) with (D.10) substituted for  $W_T$ .

$$COST_{TG} = V_{DDd}^2 x W_{BN} \frac{C_{gs0p}}{C_{gs0n}} + \frac{V_{DDs}^2}{\eta_{CP}} (W_{BN} + 4W_T). \quad (\text{D.11})$$

In Matlab, `fminsearch(@f, Y)` finds a local minimum of function `f` that takes an input vector `Y`. Unfortunately, directly using (D.11) as `f`, with inputs  $W_{BN}$  and  $x$ , does not work as desired. There is a local minimum at the desired operating point, which `fminsearch(·)` will find given sufficiently close starting conditions, but (D.10) experiences a singularity whenever  $W_{BN}\tau_{eff}g_{ds0,TG}(x) = C_T$ . Past this singularity,  $W_T$  becomes negative and (D.11) is unbounded and negative. `fminsearch(·)` will follow this curve, producing a non-physical solution.

A separate Matlab function, `fminbnd(·)`, can find the minimum of a function with a scalar input, but this optimization requires two independent variables. The approach used in this work is to distort the inputs using a well behaved function (i.e., smooth, monotonic, etc.) such that any value of the input vector `Y` produces a physically realizable (i.e.,  $W_{BN}, W_{BT}, W_T > 0$ ) switch, and then use `fminsearch(·)` with these distorted inputs. As shown in Fig. D-2,  $x$  is limited to the range where  $\tau_{eff} > 0$ . One well behaved function that takes any real number  $x$  and maps it into the bounded range  $x_{min} < x < x_{max}$  is the arctangent,

$$x = [\tan^{-1}(x') + \pi/2] \cdot \frac{x_{max} - x_{min}}{\pi} + x_{min}. \quad (D.12)$$

For a fixed  $x$  and  $g_{ds0,TG}(x)$ , the minimum value of  $W_{BN}$  is set by (D.10) as

$$W_{BN} > W_{BN,min} = \frac{C_T}{\tau_{eff}g_{ds0,TG}}. \quad (D.13)$$

There is no upper bound on  $W_{BN}$  for a physically realizable switch, leading to the distortion function

$$W_{BN} = W_{BN,min} + e^{W'_{BN}}. \quad (D.14)$$

Any real value of  $W'_{BN}$  leads to a physically realizable switch.

The optimized switch sizes are determined using `fminsearch(@f, Y)`, where `f(x', W'_{BN})` first calculates  $x$  and  $W_{BN}$  using (D.12) and (D.14) and returns the total switch cost (D.11). The result of `fminsearch` is the optimum values of  $x$  and  $W_{BN}$ , from which  $W_T$  is calculated using (D.10). The output capacitances necessary for the input buffer and digital logic fanout

Table D.1: Energy model parameters for the transmission gate sampling switch

PARAMETER	AIOP	DESCRIPTION
$C_{gs0}$	<b>P</b>	Equivalent gate capacitance of unit FET switch
$C_{off}$	<b>O</b>	Parasitic capacitance connected to $V_{in}$ when sampling switches are open
$C_{off0}$	<b>P</b>	Parasitic capacitance of unit switch when off
$C_{par0}$	<b>P</b>	Parasitic capacitance of unit switch when on
$C_{sw,sampN}$	<b>O</b>	Sampling switch load capacitance connected to $V_{DDS}$
$C_{sw,sampP}$	<b>O</b>	Sampling switch load capacitance connected to $V_{DDD}$
$g_{ds0n}(V_G, V_S)$	<b>P</b>	On conductance of unit FET switch
$W_0$	<b>P</b>	Width of unit FET

calculations are

$$C_{off} = \frac{C_{off0n}W_{BN} + C_{off0p}W_{BP}}{W_0}, \quad (D.15)$$

$$C_{sw,sampN} = \frac{W_{BN} + 4W_T}{W_0} C_{gs0n}, \quad \text{and} \quad (D.16)$$

$$C_{sw,sampP} = \frac{W_{BP}}{W_0} C_{gs0p}. \quad (D.17)$$

This approach to force convergence using Matlab's built-in functions may seem rather roundabout, but it produces good results and is presented here as only one possible solution to this multiply constrained nonlinear optimization problem. Table D.1 lists the parameters used in the transmission gate sampling model. While not specified in the table, the relevant process parameters are extracted for P and NFETs separately.



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