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#### HARNESSING NOISE TO ENHANCE ROBUSTNESS VS. EFFICIENCY TRADE-OFF IN MACHINE LEARNING

BY

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#### DISSERTATION

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## ABSTRACT

While deep nets have achieved human-comparable accuracy in various classification tasks, they fall short significantly in terms of the robustness and cost metrics. For example, tiny engineered corruptions in deep net inputs can reduce their accuracy to zero. Furthermore, deep nets also require millions of trainable parameters, resulting in significant training and inference costs. These robustness and cost challenges are well recognized today. In response, there have been a plethora of works focusing on improving either the accuracy vs. robustness trade-off, or the accuracy vs. cost trade-off. However, simultaneous consideration of accuracy, robustness, and cost metrics is largely absent today, in part, because far fewer works have explored the robustness vs. cost trade-off. This dissertation aims to fill this gap by focusing explicitly on the robustness vs. cost trade-off in the presence of data noise, as well as hardware noise. Specifically, we explore how to harness the noise in order to enhance this trade-off. We characterize and improve robustness vs. cost trade-offs across diverse problem settings, ranging from beyond-CMOS hardware implementations of machine learning (ML) classifiers to efficient training of deep nets that are robust to multiple types of corruptions in their inputs. This dissertation can be roughly divided into two part, one focusing on hardware noise and the other on data noise.

In the first part, we start by focusing on harnessing noise in spintronic hardware implementations, where the logic gates become error prone when operated at lower switching energy/delay. We propose techniques to shape the resulting hardware noise distribution and to efficiently compensate it at the system-level output. As a result, we observe  $1000 \times$  improvement in tolerance to gate-level switching error rates, while keeping the area/energy overhead of compensation circuits to as low as 15%. These robustness enhancements further enable  $3 \times$  reduction in iso-throughput energy consumption of a binary ML classifier employed for EEG-based seizure detection. Building on this work, we propose spintronic channel networks, exponential decay of spin current to efficiently realize multi-bit dot product computation. We employ error-prone nanomagnets as efficient stochastic slicers biased by spin currents proportional to the likelihood of the classification decision. We achieve  $112\times$ -to- $22.5\times$  and  $14\times$ -to- $2.5\times$  higher energy-efficiency over conventional spin-based and 20 nm CMOS designs, respectively, when realizing 10-to-100-dimensional binary classifiers. Furthermore, we also consider the impact of hardware noise originated from process variations and readout circuits in in-memory computing implementations employing non-volatile resistive crossbar arrays. Based on our analysis, we identify design configurations achieving the highest signal-to-noise ratio (SNR), and further estimate how such robustness trades off with the array energy consumption.

In the second part, we switch gears to improve the robustness vs. cost trade-off for deep nets in the presence of data noise. Specifically, we focus on the impact of adversarial perturbations in the deep nets inputs. We propose and validate the hypotheses about orientations of dominant subspaces of adversarial perturbations. We demonstrate how changes in the curvature of decision boundary of the deep nets affects the orientations of the adversarial perturbations. Based on these insights we demonstrate how shaped noise can be introduced as a feature to enhance robustness vs. cost trade-off in deep nets. Specifically, we propose shaped noise augmented processing (SNAP), a method to efficiently train deep nets that are robust to multiple types of adversarial perturbations, simultaneously. SNAP prepends a deep net with a shaped noise augmentation layer whose distribution is learned along with the network parameters using any established robust training framework. Based on extensive comparisons with nine state-of-the-art (SOTA) robust training frameworks, we show that SNAP achieves the best robustness vs. training cost trade-off. In particular, it enables  $4 \times$  reduction in the training cost compared to the SOTA approach published just this last year. Furthermore, thanks to the computational simplicity of SNAP, it is the first technique of its kind that is scalable to large datasets, such as ImageNet.

We conclude by identifying potential directions for future research.

To my parents, for all their love and support.

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# CHAPTER 1 INTRODUCTION

Machine learning (ML) systems strive to perform recognition, synthesis and policy-making tasks on the given input data. The examples of such systems include, robots using cameras to learn navigation in unseen environments, voice/chat assistants, and others. Such systems, when perfected and deployed, have potential to revolutionize multiple industries. The last five years have seen a huge surge of interest in the research and development of ML systems. It stems from the recent success of deep nets (see Fig. 1.1) in achieving human-comparable accuracy across diverse well-defined tasks, such as image recognition [1], speech recognition [2], and strategy games including Chess and Go [3].

The above accuracy metric measures the probability of given ML systems making correct decisions on unseen inputs, under idealistic conditions, *i.e.* the training data distribution is the same as that of the test data and the underlying hardware provides high precision deterministic implementation. Achieving high accuracy is necessary, but not sufficient to enable wide-scale deployment of ML systems in the real world. This is because the real world, as we all are aware, is often messy, noisy, non-ideal, and at times, even adversarial. For example, the underlying hardware itself could be erroneous due to device/circuit-level variations and noise [4, 5, 6, 7, 8]. Similarly, the inputs to ML systems could be corrupted via natural or engineered perturbations [9, 10]. Hence, it is necessary for ML systems to preserve their accuracy in the presence of noisy/non-ideal conditions. In adversarial ML research area, the term *robust accuracy* is defined as the accuracy of the classifier in the presence of engineered perturbations in its inputs. Hence, in this dissertation, we extend it to define *robustness* of a given ML system as a measure of the probability of correct decisions under *non-ideal* and *noisy* conditions. While noise is omnipresent, it can be broadly classified into two categories based on its sources:

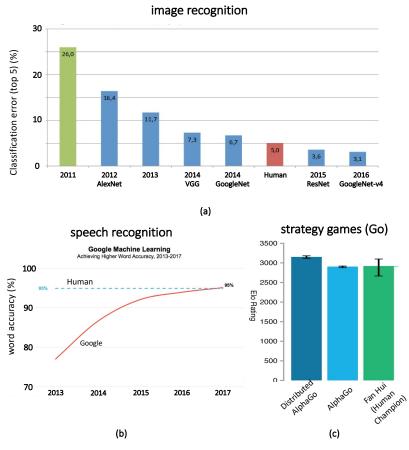


Figure 1.1: Recent success of deep nets: They have achieved state-of-the-art, human-comparable accuracy across diverse set of tasks, such as (a) image recognition [1], (b) speech recognition [2], and (c) strategy games [3].

- Data noise: this category includes natural weather-like corruptions, engineered adversarial transformations, and/or the noise in front-end sensing circuits of the ML systems.
- Hardware noise: this category includes switching errors in hardware logic gates, noise due to reduced precision and analog computation, and/or impact of process and voltage variations.

Unfortunately, human-comparable accuracy of state-of-the-art (SOTA) ML algorithms does not translate to equivalently high robustness. For example, the image recognition accuracy of deep nets gets severely lowered in the presence of common corruptions [10] (*e.g.* snow, fog, and others). Furthermore, even tiny, imperceptible perturbations in their inputs can fool deep nets to make decision errors on *every* single input in the test dataset [9, 11, 12, 13,

14]. Similarly, emerging beyond-CMOS devices suffer from increased vulnerability to hardware noise and variations [4, 8, 15, 16], causing significant drop in the inference accuracy. Hence, it is important to develop techniques that enhance robustness of ML systems, while preserving their accuracy.

The discussion of accuracy and robustness of ML systems is incomplete without the consideration of their efficiency. In this dissertation, we capture it via the cost metrics, which include both the time required to train the model (training cost), as well as energy and latency associated with producing single decision in deployment (inference cost per decision). Deep nets today require millions of trainable parameters to achieve their state-of-theart accuracy [1, 17], resulting in significant training and inference costs. This challenge is further exacerbated due to current hardware limitations such as expensive memory accesses [18], slowing down of CMOS scaling [19, 20], and increased vulnerability to process variations in both CMOS [5] and beyond-CMOS [4, 21, 8] devices.

Today both robustness and cost challenges are well-recognized. In response, a large number of works have focused on improving either accuracy vs. robustness trade-off (see Sec. 1.1), or accuracy vs. cost trade-off (see Sec. 1.2). However, simultaneous consideration of accuracy, robustness, and cost metrics is largely absent today, in part, because far fewer works have explored robustness vs. cost trade-off. This dissertation aims to fill this gap by focusing explicitly on robustness vs. cost trade-off in the presence of data noise, as well as hardware noise (see Sec. 1.3). We outline dissertation organization in Sec. 1.4.

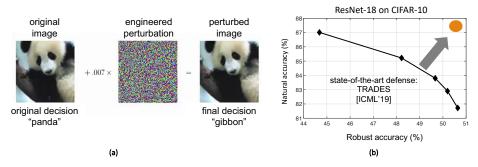


Figure 1.2: Adversarial vulnerability of deep nets: (a) example of imperceptible adversarial perturbation resulting in incorrect decision [9], and (b) trade-off between natural accuracy and robust accuracy observed for the state-of-the-art defense TRADES [22].

## 1.1 Accuracy vs. Robustness Trade-off

In spite of the high classification accuracy of deep nets [1], they are surprisingly easily fooled by corruptions in their inputs. For example, their accuracy reduces significantly in the presence of common weather-like corruptions [10], strange poses of familiar objects [23], and image rotations [24]. Furthermore, even tiny imperceptible changes in their inputs, *i.e. adversarial perturbations* [9], are sufficient to cause incorrect classification. Figure 1.2(a) shows an example of such perturbation, where despite being imperceptible, it changes the classification decision of a deep network AlexNet [9]. Similar perturbations can be found for any given deep net and any given input of that network. Furthermore, such a phenomenon has been observed across domains, *i.e.*, in image [9], speech [25], text [26], as well as, video games [27].

Multiple approaches to construct adversarial examples (attack methods) have been proposed [28, 11, 29, 12, 30, 31, 32, 33, 34]. While the basic idea is to move the input in the direction of increasing loss, stronger attacks [13, 12, 34, 30, 31] operate iteratively by perturbing the input in small increments. Almost all of these attacks succeed *deterministically*, *i.e.*, they cause classification errors for all inputs in the test dataset, against a naturally trained deep net. On the defense side, early attempts to provide adversarial defenses employed techniques such as introducing non-differentiable and/or randomized operations in the forward pass, ensemble of multiple networks [35, 36, 37, 38, 39, 40, 41]. However, such defenses were shown to be ineffective [42, 14], especially when the attacker takes explicit measures to evade that particular defense, *i.e.*, *adaptive* attacks. One example of adaptive attack is when an attacker employs expectation over transformation (EOT) to eliminate the impact of randomization in defense technique [42]. Thus, defending DNNs against adversarial attacks remains a formidable challenge partly due to a lack of in-depth understanding of the underlying cause of its vulnerability to adversarial perturbations.

Today *adversarial training* (AT) provides state-of-the-art (SOTA) empirical defense against adversarial perturbations. In AT, adversarial perturbations are computed during training to optimize a *robust* loss function [13]. After its initial success, multiple extensions and improvements in AT have been proposed [22, 43, 44, 45, 46, 47, 48, 49, 50, 51, 52, 53, 54]. In all these works, an improvement in robustness was always accompanied with some reduction in natural accuracy, *i.e.*, all of them observe a trade-off between accuracy vs. robustness [55, 56]. For example, Fig. 1.2(b) shows such a tradeoff for one of the SOTA AT framework called TRADES [22] on CIFAR-10 dataset. Notice that robust accuracy increases only at an expense of reduction in natural accuracy. Furthermore, robust accuracy is significantly (by  $\sim 40\%$ ) lower than natural accuracy even for the state-of-the-art defense. We will discuss in Sec. 1.3 how any efforts in further improving the robustness requires one to start considering robustness vs. cost trade-off.

The second source of noise is the device/circuit-level non-determinism in hardware implementations. Today, all digital CMOS implementations require deterministic switching of their component switches and gates. The problem of robustifying computation in the presence of unreliable/noisy components was posed, as early as, in 1956 by John von Neumann in [57]. He defined robust logic network as one whose output exhibits a probability of error  $p_e < 0.5$  when designed using  $\epsilon$ -noisy logic gates, *i.e.*, gates whose outputs are in error with probability  $\epsilon$ . It was further demonstrated that a robust logic network can be designed for any logic function provided  $\epsilon \leq 0.0073$ and that it is impossible to do so if  $\epsilon > \frac{1}{6}$ . Later tighter upper bounds on  $\epsilon$ were obtained in a series of papers [58, 59] culminating with those of Evans and Schulman [60]. In all these works, the overhead of additional gates required to achieve robustness was found to be impractically high, *i.e.* the cost associated with improving accuracy vs. robustness trade-off was too large.

More recently, a class of statistical error compensation (SEC) techniques was proposed to achieve robust computation in the presence of process/voltage induced delay variations in CMOS digital implementations of signal processing/ML applications [7, 61, 62]. In SEC, the hardware errors due to delay variations were compensated at the final system-level output to maintain the complexity of compensation circuits within 15%-to-20% of the main block computation [63]. As a result, up to  $16 \times$  increase in error tolerance was achieved at a marginal drop in accuracy [7]. Such robustness gains were traded-off with up to  $4 \times$  improvement in the energy-efficiency.

Finally, researchers explored modifying ML classifier training flow to enhance its robustness to hardware noise during inference. The examples include hardware-in-a-loop training of classifiers [64, 65] and hardware-aware noise injection during training [8, 16, 66, 15]. For emerging non-volatile memories, such approaches can also be combined with adaptive program-verify

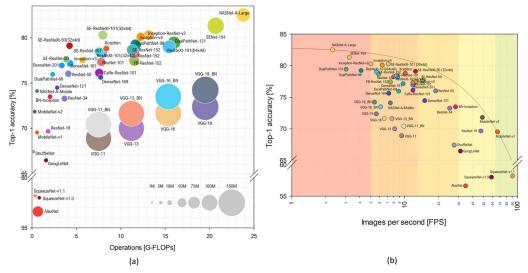


Figure 1.3: Accuracy vs. cost trade-off in deep nets [67]: classification accuracy vs. (a) number of floating-point operations and (b) images per second on NVIDIA Jetson TX1 GPU of the deep nets classifying Imagenet [68] 2012 ILSVRC dataset. The size of each ball in Fig. (a) denotes the number of parameters in the network. Millions of parameters and floating-point operations are required to achieve the SOTA accuracy.

write techniques employed to minimize the impact of write variations [21]. All these methods achieve certain improvements in the robustness to hardware noise and variations at a cost of small reduction in the accuracy.

## 1.2 Accuracy vs. Cost Trade-off

As shown in Fig. 1.1(a), deep nets achieved human-comparable accuracy in image classification tasks in 2014-15. Early proposals, such as VGGNet, were very expensive in terms of the number of network parameters and floatingpoint operations (FLOPs) required to produce a single decision. Since then, plethora of efforts have focused on improving accuracy vs. cost trade-off in deep nets as observed in Fig. 1.3(a). The proposals included both improved network architectures (*e.g.* [1, 17, 69, 70, 71, 72, 73, 74, 75]), as well as better training techniques (*e.g.* [76, 77, 78]) to achieve significant gains in accuracy vs. cost (FLOPs per inference) trade-off. The reduction in the parameter count and FLOPs also translates in improved throughput in hardware for image classification. For example, Fig. 1.3(b) shows corresponding accuracy vs. images classified per second for recently proposed deep nets running on

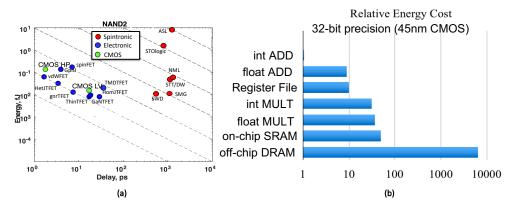


Figure 1.4: Hardware Challenges: (a) NAND2 gate switching energy vs. switching delay comparisons of the multiple beyond-CMOS devices under exploration, and their comparison with 14 nm CMOS [97, 98], (b) relative energy costs of different 32 bit operations for 45 nm CMOS implementations [18]. None of the beyond-CMOS devices on horizon significantly outperform CMOS and both on-chip and off-chip memory accesses are significantly costlier than compute operations in CMOS.

NVIDIA Jetson TX1 GPU. Researchers have also explored strategies, such as pruning [79, 80, 81, 82, 83, 84, 53, 85], aggressive quantization [86, 87, 88, 89], and neural architecture search (NAS) [90, 91, 92, 93, 94, 95, 54], to push the envelop of accuracy vs. robustness trade-off. Thanks to all these efforts, bringing the power of deep nets to the cost-constrained microcontroller-based edge devices is now a real possibility [96]. However, almost all these works do not focus on the robustness aspects discussed in Sec. 1.1. Only recently early exploration in this direction was started by combining adversarial training with pruning/NAS techniques [53, 85, 54].

All of the works discussed above improve accuracy vs. cost trade-off within the constraints of the current hardware. In order to push their envelope further, one needs to address today's hardware limitations. For instance, as the channel lengths continue to reduce beyond a few tens of nanometers, the energy and delay reductions due to CMOS scaling have stagnated. Furthermore, while multiple beyond-CMOS devices are under exploration, none of them significantly outperform CMOS in terms of gate-level switching energy and delay [97, 98], as evident in Fig. 1.4(a). Even in CMOS implementations, accessing data from memory is more expensive than doing computations on that data [18], as shown in Fig. 1.4(b). For example, fetching 32 bit word from register file or on-chip SRAM is almost  $10 \times$  more energy-expensive

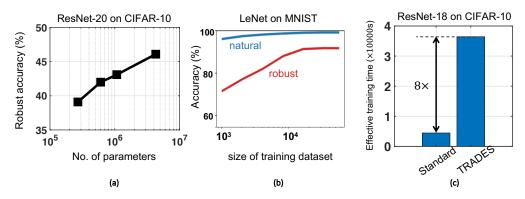


Figure 1.5: Interwoven robustness and cost challenges: Robust accuracy vs. (a) no. of model parameters [99] for ResNet-20 network trained on CIFAR-10 dataset, (b) natural and robust accuracy vs. size of training dataset for MNIST dataset [100], and (c) comparison of effective training time for standard training and TRADES adversarial training [22].

than 32 bit int ADD operation. Furthermore, off-chip DRAM accesses are almost  $10 \times$  more energy-expensive than even iso-precision int MULT. This challenge is particularly exacerbated for deep net implementations due to their large parameter requirements. In the Sec. 1.3, we will discuss how the current approaches addressing these challenges, in fact, requires one to focus on robustness vs. cost trade-off due to hardware noise.

## 1.3 Dissertation Contribution: Improving Robustness vs. Cost Trade-off

As discussed in Sec. 1.1 and Sec. 1.2 the accuracy vs. robustness trade-off and accuracy vs. cost trade-off are primarily explored independently today. However, they start getting interwoven as one attempts to design ML implementations at the limits of accuracy, robustness, and efficiency. For instance, while pushing the envelop of accuracy vs. robustness trade-off via AT, one begins to run into prohibitive increase in training and inference costs as shown in Fig. 1.5, *i.e.* one needs bigger models [99, 104] (Fig. 1.5(a)), larger training dataset [100, 105] (Fig. 1.5(a)), and at times, even  $8 \times$ -to- $10 \times$  more complex training algorithms [22, 106, 107]. On the other hand, quest for achieving leaps of energy-efficiency in hardware requires addressing the robustness challenge in the presence of hardware noise [108, 65, 5]. For instance, prominent beyond-CMOS devices, such as RRAM and spin-based devices,

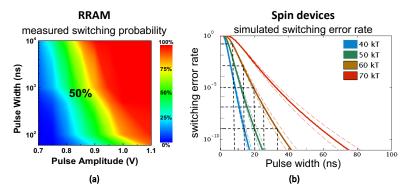


Figure 1.6: Intrinsically non-deterministic nature of emerging beyond CMOS devices: (a) measured switching probability vs. pulse width vs. pulse amplitude trade-off for RRAM devices [101], (b) simulated switching error rate vs. pulse width trade-off for spin-based devices with different energy barriers [102, 103].

are inherently non-deterministic in nature [102, 101, 103]. Their switching error rates increase with the reduction in their switching energy/delay as shown in Fig. 1.6. Similar trends are also observed in one of the promising approach called *in-memory computing* (IMC), where one strives to embed computation within the bitcell array (BCA) and its peripherals in order reduce the memory access costs. Due to severe area constraints in the BCA, such in-memory computation often needs to happen in the analog domain along bitlines (BLs) by using change in BL voltage  $\Delta V_{BLB}$  to represent multibit information (see *e.g.* Fig. 1.7(a)). With decreasing BL discharge, a key knob for reducing energy cost of SRAM read, the variations in  $\Delta V_{BLB}$  increase as shown in Fig. 1.7(b) [65]. It adversely affects the classifier accuracy (also shown in Fig. 1.7(b) [65]) indicating the need for improving robustness against hardware noise to achieve maximum energy benefits.

In summary, improving the accuracy vs. robustness trade-off in the presence of data noise exacerbates the need for cost-efficiency, while one needs to achieve robustness against hardware noise to further enhance the accuracy vs. cost trade-off. To address this predicament, we believe that one needs to focus explicitly on improving the robustness vs. cost trade-off. Specifically, we ask following fundamental questions:

How should the noise be managed in order to enhance robustness vs. cost trade-off in ML systems? Is noise always a problem that should be mitigated? Can it be exploited as a feature?

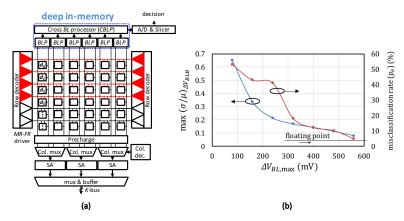


Figure 1.7: Need for robustness against hardware noise in in-memory computing: (a) deep in-memory architecture (DIMA), and (b) variations in  $\Delta V_{BLB}$ and their impact on classifier accuracy as a function of maximum BL discharge, a key knob controlling energy consumption.

This dissertation attempts to address these questions in diverse problem settings across the design stack. Based on our work, we find that noise need not always be mitigated, but the robustness vs. cost trade-off in the presence of both hardware noise and data noise can be enhanced via *noise shaping*. For example, in Chapter 2, we show how hardware noise distribution in spintronic digital implementations can be shaped to achieve robust computation even in the presence of error-prone logic gates. As an another example, in Chapter 7, we demonstrate how *shaped noise* can be specifically introduced in the deep nets to achieve high adversarial robustness at significantly lower training cost.

## 1.4 Dissertation Organization

Our contributions in this dissertation are organized as follows:

Chapter 2 explores how the noise distribution in spin-based digital implementations can be shaped by exploiting inherent trade-off between error rate, energy, and delay for spin-based logic gates. Such noise shaping enabled efficient error compensation via a Shannon-inspired model of computation. Resulting classifier implementation is  $1000 \times$  more tolerant to gate-level switching error rate compared to the conventional implementation. These robustness improvements are translated into energy-efficiency gains.

**Chapter 3** demonstrates how the impact of spin device noise is fundamentally changed via novel circuit design and algorithmic techniques to achieve energy-efficient inference implementations. In particular, this chapter presents *spin channel networks (SCN)* – a novel spin-based circuit design approach that exploit exponential decay of spin current to efficiently realize multi-bit dot product computation. Furthermore, Adaptive Boosting (AdaBoost) framework is employed to design multiple isolated tiny spin channel networks (t-SCNs) that work in unison to solve an arbitrary binary classification task. Such boosted t-SCNs achieve  $112 \times -to-22.5 \times$  and  $14 \times -to-2.5 \times$ higher energy-efficiency over conventional spin-based and 20 nm CMOS designs, respectively, when realizing 10-to-100-dimensional binary classifiers.

Chapter 4 proposes an MRAM-based deep in-memory architecture to achieve multi-bit matrix-vector multiplication (MVM) within a single read operation of MRAM bitcell array (MRAM-BCA). The MRAM-BCA peripheral circuits are modified to potentially achieve such multi-bit computation at  $20 \times$  and  $10 \times$  lower energy and delay, respectively, compared to digital MRAM implementation. The robustness challenges that emerge as a byproduct of achieved energy benefits are further characterized, and techniques to address them are discussed.

**Chapter 5** presents signal-to-noise ratio (SNR) analysis for in-memory computing in resistive crossbar arrays. Specifically, we show how the circuitlevel specifications of peripherals affect SNR as a function of array size, sensing circuit impedance, as well as energy per operation for PCM, MRAM, and RRAM. The noise sources considered are mismatch in the conductance and input digital-to-analog converters (DACs), as well as, clipping and quantization noise in the readout circuits. We conclude by proving some design guidelines for appropriate noise budgeting in order to achieve accurate inmemory matrix vector multiplication in the in-memory crossbar arrays.

**Chapter 6** aims to develop deeper understanding of geometric orientations of adversarial perturbations of deep nets via subspace analysis for image classification tasks. Specifically, we propose and validate the hypotheses about orientations of dominant subspaces of adversarial perturbations. We demonstrate how changes in the curvature of decision boundary of the deep nets affects the orientations of the adversarial perturbations. The hypotheses and insights developed in this chapter lead to noise shaping and augmentation techniques proposed in Chapter 7.

**Chapter 7** explores how shaped noise can be employed as to enhance adversarial robustness vs. cost trade-off in deep nets based on the insights from Chapter 6. Specifically, we propose shaped noise augmented processing (SNAP), a method to efficiently train deep nets that are robust to multiple types of adversarial perturbations, simultaneously. SNAP prepends a deep net with a shaped noise augmentation layer whose distribution is learned along with the network parameters using any established robust training framework. Based on extensive comparisons with nine state-of-theart (SOTA) robust training frameworks, we show that SNAP achieves the best robustness vs. training cost trade-off. Furthermore, thanks to the computational efficiency of our approach, we report for the first time ResNet-50 (ResNet-101) networks on ImageNet that achieve > 30% robust accuracy against the union of  $(\ell_{\infty}, \ell_2, \ell_1)$  adversarial perturbations.

**Chapter 8** outlines a few potential extensions and future research directions.

## CHAPTER 2

## ROBUST SPINTRONICS VIA SHANNON-INSPIRED APPROACH

## 2.1 Overview

As we discussed in Chapter 1, CMOS scaling is slowing down and none of the beyond-CMOS devices are yet able to significantly outperform CMOS [97, 98]. Spin-based computational devices built with nanomagnets and spinpolarized transport have emerged as a viable beyond CMOS option, due to their following favorable attributes: (i) non-volatility, (ii) higher logical efficiency, and (iii) high integration density and compatibility with the state-ofart back-end electronics manufacturing processes. These devices are a subset of the beyond-CMOS devices which include devices based on electron spin [109, 103] and magneto-electric [110, 111] phenomena. However, spin-based devices are not competitive to CMOS [98], in terms of switching energy and delay, due to their high energy-delay requirements to achieve deterministic switching [102, 112, 113, 114]. As switching energy or delay is reduced, their switching error probability increases, rendering them incompatible with the required determinism of the digital logic.

In this chapter, we demonstrate how spin-based digital implementations of ML classifiers can be made robust to a significant increase in the switching error probability of their component logic gates. We employ the Shannon-inspired model of computation [5] to enhance such gate-level switching error tolerance. In the Shannon-inspired framework, hardware errors are engineered and then efficiently compensated via the introduction of tailored redundancy, in the spirit of Shannon's theory for communications [115]. The detailed contributions of this chapter are as follows:

• We characterize the  $\epsilon$ -energy-delay trade-off for ASL gates to enable non-uniform  $\epsilon$  assignments across logic gates.

- We propose logic-level path delay reallocation techniques to assign appropriate error rates to individual gates such that the resulting output error distributions are shaped to facilitate error compensation.
- We propose a novel maximum likelihood (ML) error compensation scheme that exploits these shaped output error statistics to compensate the errors efficiently.
- We demonstrate a 1000× higher average error rate tolerance and a 3× lower energy-per-decision for an ASL-based digital support vector machine (SVM) implementation, while maintaining its system-level classification accuracy.

## 2.2 Background

#### 2.2.1 All Spin Logic Device

Figure 2.1(a) shows a diagram of an ASL inverter. It consists of two nanomagnets separated by a conducting channel of length L. The input magnet  $(M_{\rm in})$  polarizes the supply current passing through it. This creates a spin concentration gradient and propagates a spin current in the channel of length L. This spin current, in turn, exerts a torque on the magnetization of the output magnet  $(M_{\rm out})$  forcing it to switch. Since the magnets are non-volatile, they retain the magnetization vector state when the supply current is switched off.

Electrical current in the order of  $10 \,\mu A$ -to- $100 \,\mu A$  is required to generate sufficient spin current to switch the output magnet. Since the nanomagnets and the spin channel are metallic, the equivalent electrical resistance across the nanomagnet-channel stack is small (few  $\Omega$ s), enabling these devices to operate at ultra-low supply voltages. However, the electrical current through the input nanomagnet flows irrespective of output activity, causing high static energy consumption. The nanomagnets, being non-volatile, retain the magnetization vector state even when the supply current is switched off. Hence, researchers propose to clock these devices via a MOSFET [116, 117], operating in the linear region, which acts as a switch turning ON the ASL device only when it needs to compute as shown in Fig. 2.1(a). The ON

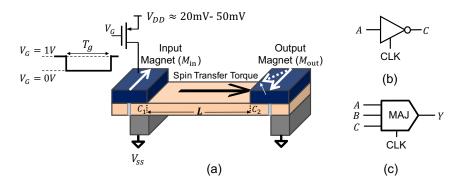


Figure 2.1: All Spin Logic (ASL) (a) diagram of clocked ASL inverter gate [118, 117], (b) clocked ASL inverter symbol, and (c) clocked ASL 3-majority gate symbol.

duration  $T_g$  of clock can be externally controlled for each gate. Thus, the energy consumption of the clocked ASL gates is completely determined by  $T_g$  and the ON current of the gating MOSFET. Figure 2.1(b) and 2.1(c) show the logical symbols for clocked ASL inverter and 3-majority gate, respectively. Pajouhi *et al.* [116] proposed to share a single MOSFET across multiple nanomagnets by electrically stacking their supply terminals in series to significantly amortize the clock pulse generation and MOSFET switching overheads. In this work, we assume such amortization described in [116] and focus on the impact of gate-level switching errors on the final output.

#### 2.2.2 Support Vector Machine

Linear SVM [119] is a simple and popular machine learning algorithm for binary classification. The SVM learns a hyperplane to separate the training feature vectors into two regions, each corresponding to one class, as shown below:  $\hat{z}=1$ 

$$\mathbf{w}^T \mathbf{x} + b \underset{\hat{z}=-1}{\overset{z=-1}{\gtrless}} 0$$

where  $\mathbf{w}$  and b denote the trained weight vector and bias representing the separating hyperplane, respectively,  $\mathbf{x}$  denotes the *N*-dimensional input feature vector, and  $\hat{z}$  denotes the predicted label. If the true label is denoted by z, the accuracy of SVM is given by the probability of classification error  $p_e = \Pr{\{\hat{z} \neq z\}}$ , which can be empirically estimated for a given dataset.

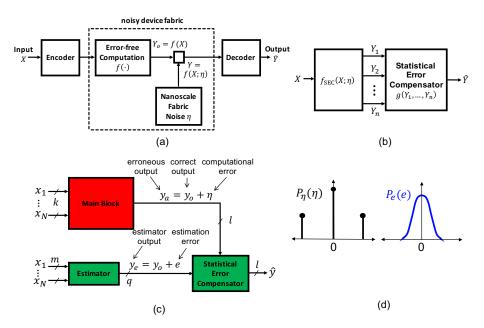


Figure 2.2: The Shannon-inspired model of computation: (a) model, (b) statistical error compensation (SEC), (c) algorithmic noise tolerance (ANT), a special case of SEC, where the error compensator combines two unreliable outputs  $y_a$  and  $y_e$ , (d) illustrative distributions of computational error  $\eta$ , estimation error e that lead to a low-complexity and accurate error compensator.

#### 2.2.3 Shannon-inspired Model of Computation

The Shannon-inspired model of computation [5] (Fig. 2.2(a)) comprises an encoder, a noise-free computation of the desired correct output  $Y_o = f(X)$  being corrupted by noise in nanoscale fabrics parametrized by variable  $\eta$  to generate the observed output  $Y = f(X; \eta)$  of the error-prone device fabric (the channel), followed by the decoder that recovers the corrected output  $\hat{Y}$ . In Fig. 2.2(a), all variables  $(X, Y_o, \eta, Y, \hat{Y})$  are random variables. In this chapter, we use capitals to denote random variables and small letters to denote their particular instance. For example, Y denotes a random variable, while y denotes a specific value of Y.

Statistical error compensation (SEC) (Fig. 2.2(b)), one class of the design techniques within the Shannon-inspired framework [5, 120], introduces a statistical error compensator block as a decoder, which combines multiple unreliable outputs  $Y_1, \ldots, Y_n$  to compute corrected output  $\hat{Y}$ . Algorithmic noise tolerance (ANT) (Fig. 2.2(c)) is a special case of SEC where the error compensator combines two unreliable outputs  $y_a$  and  $y_e$ . ANT consists of a main block designed using unreliable/noisy device fabric accounts for 85%- 90% of total gate count complexity. It strives to compute correct output  $y_o$ , but ends up computing  $y_a$  due to the unreliability of the underlying device fabric. ANT augments the main block with a low complexity estimator that computes an estimate  $y_e$  of the correct output  $y_o$ . Under the assumption of additive noise model, the main block and estimator outputs are described as follows:

$$y_a = y_o + \eta \tag{2.1}$$

$$y_e = y_o + e \tag{2.2}$$

where  $\eta$  is a system-level hardware error observed at the main block output, and e is the estimation error incurred due to inherent lower complexity of the estimator.

The estimator and the error compensator are designed using reliable, and hence energy-inefficient, circuits, constituting the error compensation overhead in ANT. Hence, their combined complexity (in terms of gate count) needs to be significantly ( $\approx$  5-to-10×) smaller than the main block. Previously, it has been shown that [7, 121, 122] the complexity of the error compensator can be reduced by shaping the distributions of  $\eta$  and e,  $P_{\eta}(\eta)$ and  $P_e(e)$ , respectively, to be disparate from each other as shown in Fig. 2.2(b) and 2.2(c). In particular, a dense  $P_e(e)$  is realized by introducing a reduced-precision estimator, while a sparse  $P_{\eta}(\eta)$  is realized by permitting MSB errors in the LSB-first architectures [121, 123, 63, 122]. Various design techniques to reduce the overhead of the estimator and the error compensator have been proposed [63, 124, 125, 123].

#### 2.2.4 Mutual Information (MI)

The mutual information (MI) I(X;Y) between two random variables X and Y quantifies the amount of information conveyed about X by knowing the value of Y, and vice versa. The MI I(X;Y) is defined as:

$$I(X;Y) = H(X) - H(X|Y) = H(Y) - H(Y|X)$$
(2.3)

where H(X) and H(X|Y) denote the entropy of X and conditional entropy of X given Y, respectively. The entropy H(X) of a random vari-

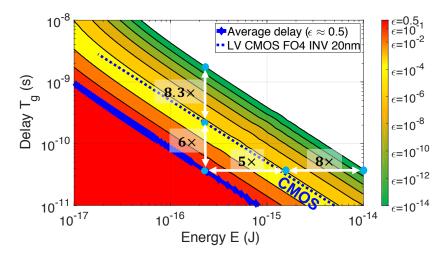


Figure 2.3: Trade-off between switching error rate  $\epsilon$ , switching energy  $E_g$ , and switching delay  $T_g$  for a clocked ASL inverter gate.

able X quantifies the uncertainty about the value of X, and is a function of its probability distribution. In this chapter, we use MI metric to show that the Shannon-inspired model of computation (Fig. 2.2) enhances the MI  $I(Y_o; Y_a, Y_e)$ , thereby enabling an accurate recovery of  $y_o$  from  $y_a$  and  $y_e$ .

## 2.3 Modeling Stochasticity of ASL Devices

In this section, we develop a gate-level model to capture the inherent devicelevel stochasticity of ASL at the circuit and architecture level. Even after receiving supply current  $I_{\rm on}$  (>  $I_{\rm crit}$ ) at the input nanomagnet, the output nanomagnet of the ASL gate may not switch due to the presence of Langevin thermal noise [102, 112, 113, 114], where  $I_{\rm crit}$  denotes the minimum current required for nanomagnetic switching. In this chapter, we refer to this probabilistic event as *switching error*, and its probability  $\epsilon$  as the *switching error rate*. In [102], an analytical expression for  $\epsilon$  was derived by employing the Fokker-Planck equation for magnetization vector switching dynamics governed by the fundamental LLG equation and was validated against Landau-Lifshitz simulations of a macrospin including appropriate thermal field. This analysis indicates a gate-level trade-off between switching error rate  $\epsilon$ , the switching energy  $E_g$ , and the switching delay  $T_g$  of ASL gates.

Figure 2.3 shows the iso-error rate delay vs. energy contours of an ASL inverter at various error rates. As expected, the error rate decreases with

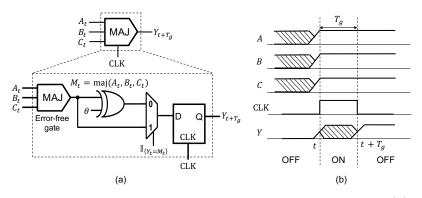


Figure 2.4: The modified  $\epsilon$ -noisy gate model for clocked ASL: (a) a gatelevel schematic emulating the stochastic behavior of a non-volatile, clocked ASL 3-majority gate, and (b) timing diagram illustrating the phase where the gate is ON and OFF.

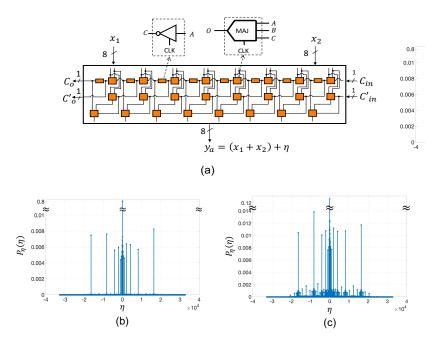


Figure 2.5: RCA with gate-level uniform error rate  $\epsilon$  assignment operating at total delay of 1.24 ns: (a) schematic of 8 bit RCA showing all gates operating at  $\epsilon = \epsilon_{\rm cp-avg} = 10^{-2}$ , and error distribution  $P_{\eta}(\eta)$  for a 15 bit RCA (b) when  $\epsilon = \epsilon_{\rm cp-avg} = 10^{-2}$ ,  $E_{\rm RCA15} = 90$  fJ, and (c) when  $\epsilon_{\rm cp-avg} = 10^{-1}$ ,  $E_{\rm RCA15} = 60$  fJ, where  $E_{\rm RCA15}$  denotes total switching energy of 15 bit RCA.

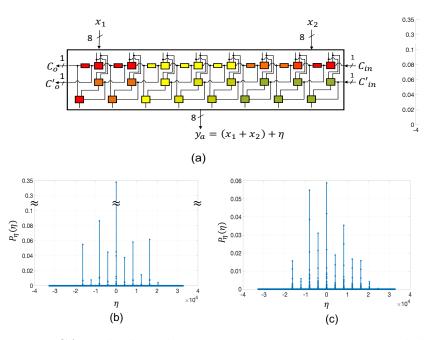


Figure 2.6: RCA with shaped error statistics operating at total delay of 1.24 ns: (a) schematic of an 8 bit RCA illustrating spatial distribution of gatelevel  $\epsilon$  after applying PDB and PDR, and error distribution  $P_{\eta}(\eta)$  for a 15 bit RCA (b) when  $\epsilon_{\text{cp-avg}} = 10^{-2}$ ,  $E_{\text{RCA15}} = 90$  fJ, and (c) when  $\epsilon_{\text{cp-avg}} = 10^{-1}$ ,  $E_{\text{RCA15}} = 60$  fJ. The colors in (a) approximately convey the error rates of the gates as per the color code in Fig. 2.3.

increasing energy or delay. In fact, when  $I_{\rm on} \gg I_{\rm crit}$ , the expression for  $\epsilon$  [102] can be simplified via Taylor series approximation to:

$$\epsilon(E_g, T_g) = \beta \exp(-\zeta \sqrt{E_g T_g}) \tag{2.4}$$

where  $\beta$  and  $\zeta$  are device-dependent constants A 3-majority ASL gate operates with error rate of  $\epsilon(E_g, T_g)$  if all its inputs are equal, and with higher error rate of  $\epsilon(\frac{E_g}{3}, T_g)$  otherwise. In this work, we conservatively upper-bound the error rate of 3-majority gate to  $\epsilon(\frac{E_g}{3}, T_g)$ . Equation (2.4) explains the observed linearity of the contours at higher values of  $E_g$  or  $T_g$  in Fig. 2.3. We further note that ASL inverter consumes  $8 \times$  more energy compared to 20 nm CMOS FO4 inverter [103] at  $\epsilon = 10^{-14}$  and at identical switching delays. Hence, ASL-based conventional digital architectures remain non-competitive with respect to present day CMOS. As  $\epsilon$  is increased beyond 1%, the ASL inverter becomes more energy-efficient than CMOS, demonstrating the potential for achieving energy-efficiency, if one can tolerate such high gate-level error rates while maintaining final system-level accuracy. We develop a modified  $\epsilon$ -noisy gate model (Fig. 2.4(a)) to describe a clocked ASL gate, which comprehends its underlying stochastic behavior, while being sufficiently abstract to permit the design and analysis of complex ASL networks. The modified  $\epsilon$ -noisy gate model captures: (a) the logic-level manifestation of device-level stochasticity, (b) the input dependence of ASL errors due to the non-volatility of the nanomagnets, i.e., the ASL gate makes an error only when the output nanomagnet fails to switch when it should, implying a dependence of the error event on the input data, and (c) the role of the CLK terminal in the gate operation.

Figure 2.4(b) shows the timing diagram for the modified  $\epsilon$ -noisy model. The Boolean inputs A, B, and C are applied at time t. The ASL gate generates its output Y at time  $t + T_g$  where  $T_g$  is the switching delay assigned to the ASL gate. The model comprises of an ideal noise-free Boolean gate whose output  $M_t = \text{maj}\{A_t, B_t, C_t\}$  is EXORed with a Bernoulli random variable  $\theta$  with parameter  $\epsilon$ , i.e.,  $\Pr\{\theta = 1\} = \epsilon$ . The output selector (implemented using a multiplexer in Fig. 2.4(b)) computes the final output  $Y_{t+T_g}$  by choosing either the output of the EXOR gate  $M_t \oplus \theta$  or the error-free output  $M_t$ . The D flip-flop models the non-volatility, i.e., the ability to retain the output when CLK = 0. The EXOR gate output is chosen only if  $Y_t \neq M_t$ , capturing the fact that the switching error can occur only if the output nanomagnet is required to switch.

## 2.4 Shannon-inspired ASL Architecture

In this section, we describe how the Shannon-inspired approach can be applied to clocked ASL networks to increase their tolerance to switching errors. In Sec. 2.4.1, we propose path delay reallocation techniques that exploit the gate-level trade-off between  $\epsilon$ ,  $E_g$ , and  $T_g$  to shape the output error statistics and thereby ease error recovery. In Sec. 2.4.2, we propose a novel fusion block architecture to compensate for the switching errors.

#### 2.4.1 Shaping Error Statistics

In clocked digital ASL networks, the random switching errors occur at the output of every logic gate as modeled in Sec. 2.3. The impact of such gate-

level errors accumulates as the input propagates to the final output. For example, consider a clocked ASL-based 8 bit ripple carry adder (RCA) consisting of all ASL gates operating at identical switching delay  $T_g$ , switching energy per nanomagnet  $E_g$ , and hence, identical  $\epsilon(E_g, T_g)$  as shown in Fig. 2.5(a). The resulting distribution  $P_\eta(\eta)$  of output error  $\eta$  for a 15 bit RCA is dense as shown in Fig. 2.5(b) and 2.5(c) for  $\epsilon(E_g, T_g) = 10^{-2}$ and  $\epsilon(E_g, T_g) = 10^{-1}$ , respectively. Brute force compensation of the errors having such distributions can be computationally expensive as discussed in Sec. 2.4.2. We propose error statistics shaping techniques to impose a structure on  $P_\eta(\eta)$  to reduce the complexity of error compensation.

We exploit the error rate, energy, and delay trade-off of the clocked ASL gates (shown in Fig. 2.3) to shape the distribution of error  $\eta$ . In particular, we control the gate-level switching delay via clock pulse width modulation as described in Sec. 2.2.1 [117, 116]. Exploiting this degree-of-freedom, we propose two logic-level delay assignment steps, namely path delay balancing (PDB) and path delay redistribution (PDR). We begin with a logic gate network with all gate delays equal to  $T_g$ . Thus, the critical paths are those with the maximum number of gates  $N_{\rm cp}$  and therefore have the path delay  $T_{\rm cp} = T_g N_{\rm cp}$ . In PDB and PDR steps, the gate delays are reassigned at a constant switching energy (per nanomagnet) of  $E_g$  (moving vertically in Fig. 2.3) and at a constant throughput (identical critical path delay  $T_{\rm cp}$ ) as follows.

#### PDB

In PDB, delays of gates lying on the shorter paths are increased, at a constant energy  $E_g$ , making every gate to lie on one or more critical paths. Thus, PDB reduces error rate of gates on shorter paths, while leaving the original critical path unaltered, now containing gates with highest error rates.

#### PDR

In PDR, the gates delays along all critical paths are further redistributed to further enhance the sparsity of  $P_{\eta}(\eta)$ , while keeping their path delay constant. In particular, the delays of the few gates in the middle of the critical path are increased (lowering  $\epsilon$ ) at the expense of the reduction in the delays (increasing

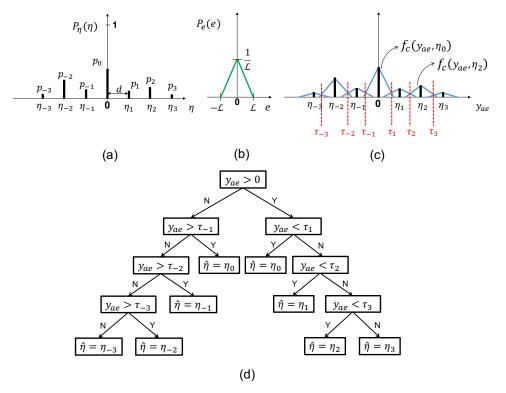


Figure 2.7: Maximum likelihood (ML) error compensation: (a) illustrative  $P_{\eta}(\eta)$  consisting of seven distinct peaks, (b) illustrative  $P_e(e)$ , (c) corresponding  $f_c(y_{ae}, \eta_i)$  defined in (2.8), and (d) TreeCompensator, the resulting ML error compensator having a decision-tree structure.

 $\epsilon$ ) of the gates lying at the beginning and at the end of the critical path. Such delay redistribution increases the error rates of the top few MSBs and bottom few LSBs, while reducing the error rates of the other bits in the middle. Doing so results in increased probability of errors having extreme magnitudes (both very high and very low), leading to a highly sparse  $P_{\eta}(\eta)$ .

We define the average device error rate of the clocked ASL network as  $\epsilon_{cp-avg} = \epsilon(E_g, T_{cp-avg})$ , where  $T_{cp-avg} = \frac{T_{cp}}{N_{cp}}$ . Note:  $T_{cp-avg} = T_g$ , when all gates on the critical path have equal delay. Figure 2.6(a) illustrates the spatial distribution in gate-level switching error rates (employing the color code from Fig. 2.3) for an 8 bit clocked ASL-based RCA after applying both PDB and PDR. The resulting  $P_{\eta}(\eta)$  for a 15 bit RCA subject to PDB and PDR is shown in Fig. 2.6(b) and (c) for  $\epsilon_{cp-avg} = 10^{-2}$  and  $\epsilon_{cp-avg} = 10^{-1}$ , respectively. Compared to the distributions in 2.5(b) and (c), the distributions in 2.6(b) and (c) are sparse, i.e., they have distinct well-separated peaks with relatively smaller spread around them.

Next, we show that error statistics shaping via PDB and PDR preserves the information in the erroneous output  $y_a$  about the correct output  $y_o$ , which can be quantified via the MI  $I(Y_a; Y_o)$ . We empirically estimate  $I(Y_a; Y_o)$  for the 15 bit RCA example in Fig. 2.5 and Fig. 2.6. For an error-free RCA,  $I(Y_a; Y_o) = 13.98$  bits, which drops to 6.18 bits, with all gates are operating at an identical error rate of  $\epsilon_{cp-avg} = 10^{-1}$ . The resulting  $P_{\eta}(\eta)$  in Fig. 2.5(c) is dense. The shaped error statistics in Fig. 2.6(c) enhances MI  $I(Y_a; Y_o)$ to 11.15 bits. Noted that there exist multiple methods of shaping  $P_{\eta}(\eta)$  to increase the MI. Furthermore, a high value of  $I(Y_a; Y_o)$  only guarantees the existence of an error compensation scheme to reliably recover  $y_o$  from  $y_a$ . However, such scheme need not be efficient. In Sec. 2.4.2, we derive a nearoptimal low-complexity error compensation scheme that exploits the sparsity of  $P_{\eta}(\eta)$ .

#### 2.4.2 Maximum Likelihood (ML) Error Compensator

The role of the fusion block in SEC is to compute the estimate  $\hat{y}$  of the correct output  $y_o$ , as a function of two error-prone observations  $y_a$  and  $y_e$  (see Fig. 2.2(a)). One approach to make  $\hat{y}$  a good estimate of  $y_o$  is to choose  $\hat{y}$  such

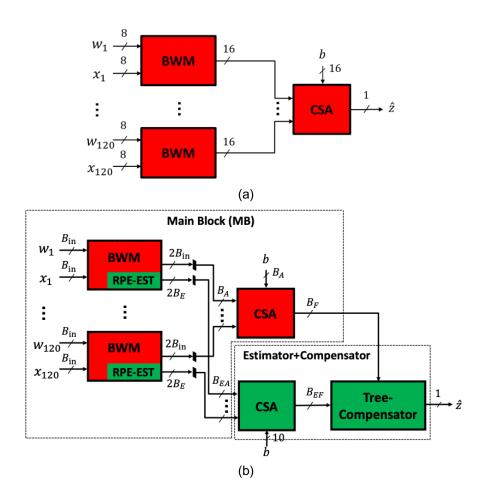


Figure 2.8: Digital clocked ASL-based 120-dimensional SVM classifiers: (a) conventional serial architecture with uniform delay assignments, (b) Shannon-inspired architecture.

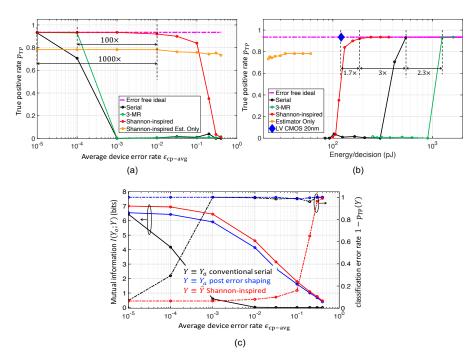


Figure 2.9: Accuracy vs. energy, error rate trade-off for different digital clocked ASL-based 120-dimensional SVM classifier implementations operating at a fixed decision delay and  $p_{FA}$  of 1%: (a) True positive rate  $p_{TP}$  vs. average device error rate  $\epsilon_{cp-avg}$ , (b)  $p_{TP}$  vs. total classifier energy per decision, (c) mutual information (MI)  $I(Y_o; Y)$  and corresponding classification error rate  $(1 - p_{TP})$  vs.  $\epsilon_{cp-avg}$  curves for serial architecture (black), and after shaping its error  $\eta$  statistics (blue), and Shannon-inspired architecture (red).

that it maximizes the likelihood of the observations  $y_a$  and  $y_e$  as follows:

$$\hat{y} = \arg\max_{y} P_{Y_a, Y_e|Y_o} \left\{ Y_a = y_a, Y_e = y_e \middle| Y_o = y \right\}$$
(2.5)

where  $P_{Y_a,Y_e|Y_o}$  denotes the likelihood of  $Y_a$  and  $Y_e$  given  $Y_o$  and y denotes a free variable in the maximization that is swept over the range of possible values of correct output  $y_o$ . Thus,  $\hat{y}$  is a maximum likelihood (ML) estimate of  $y_o$ . In general, it can be computationally expensive to compute and maximize  $P_{Y_a,Y_e|Y_o}$ . However, the error statistics shaping described in Sec. 2.4.1 significantly reduces the computation of the ML estimate  $\hat{y}$  as shown next.

Noting the independence of  $\eta$  and e conditioned on  $Y_o$  in (2.5), we get,

$$\hat{y} = \arg\max_{y} P_{\eta}(y_a - y) P_e(y_e - y)$$
(2.6)

We employ parametric models for  $P_{\eta}(\eta)$  and  $P_{e}(e)$  [123] as shown in Fig.

2.7(a) and 2.7(b), respectively, to simplify (2.6) to:

$$\hat{y} = y_a - \hat{\eta} \tag{2.7}$$

with  $\hat{\eta}$  given as:

$$\hat{\eta} = \arg \max_{\eta_i} \left[ \underbrace{p_i \mathbb{1}_{\{\eta_i - \mathcal{L} < y_{ae} < \eta_i + \mathcal{L}\}} f_e(-y_{ae} + \eta_i)}_{f_c(y_{ae}, \eta_i)} \right]$$
(2.8)

where  $y_{ae} = y_a - y_e = \eta - e$ ,  $\Pr\{\eta = \eta_i\} = p_i$ ,  $\min_{i,j} |\eta_i - \eta_j| = d$ ,  $\Pr\{|e| < \mathcal{L}\} = 1$ , and  $f_e$  denotes a functional description of  $P_e$  when  $|e| < \mathcal{L}$ .

Given  $y_a$ ,  $y_e$ , a brute-force computation of the ML estimate  $\hat{y}$  requires evaluating (2.7) by calculating RHS of (2.8) for every  $\eta_i$ , and selecting  $\eta_i = \hat{\eta}$ that maximizes it. Figure 2.7(c) illustrates plots of  $f_c(y_{ae}, \eta)$  as a function of  $y_{ae}$  for all values of  $\eta$ . It can be observed that  $\hat{\eta}$  can be approximately computed via comparisons of  $y_{ae}$  with thresholds  $\tau_i$ s. Thus, the ML error compensator has a decision tree structure as shown in Fig. 2.7(d), and is henceforth referred to as a TreeCompensator. The thresholds  $\tau_i$ s in the TreeCompensator are a function of error distributions  $P_{\eta}$  and  $P_e$ . For a given implementation, these distributions can be characterized once during simulations, or one-time calibration phase of the prototype chip. Once the thresholds are computed offline and stored, the TreeCompensator can be implemented efficiently using only a few subtracters.

## 2.4.3 Digital Clocked ASL-based Dot Product Implementations

Figure 2.8(a) shows the conventional serial architecture of a 120-dimensional SVM classifier. It employs 8 bit signed Baugh Wooley multipliers (BWM) and a carry save adder (CSA). All gates in this architecture operate at identical error rates. The Shannon-inspired architecture in Fig. 2.8(b) employs the conventional serial architecture as the main block (MB), and applies PDB and PDR to shape its output error distribution. Since PDB and PDR techniques make some gates operate at a lower error rate, few reliable intermediate signals in BWMs can be employed as the estimates of the BWM outputs indicated via green reduced-precision embedded estimator (RPE-EST)

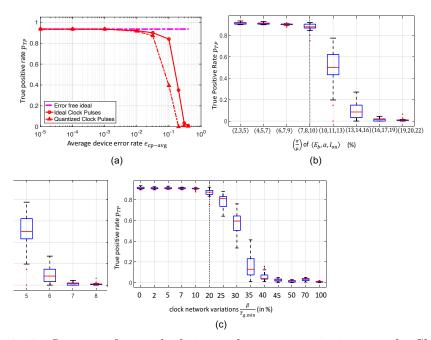


Figure 2.10: Impact of non-idealities and process variations on the Shannoninspired implementation: (a)  $p_{\rm TP}$  vs  $\epsilon_{\rm cp-avg}$  trade-off for Shannon-inspired implementation having 46 distinct clock pulse widths, (b)  $p_{\rm TP}$  box plot for different levels of static within-die process variations measured in terms of  $\frac{\sigma}{\mu}$  $E_b$ ,  $\alpha$ , and  $I_{\rm on}$  for Shannon-inspired implementation having 46 distinct clock pulse widths, and (c)  $p_{\rm TP}$  box plot as a function of extent of dynamic clock network variations  $\frac{\beta}{T_{\rm g,min}}$  for the Shannon-inspired implementation having 46 distinct clock pulse widths, and  $\frac{\sigma}{\mu}$  of  $E_b$ ,  $\alpha$ , and  $I_{\rm on}$  set at 4%, 5%, and 7%, respectively.

blocks in BWMs, similar to techniques discussed in [63] to reduce estimator overhead. The additional overhead consists of a CSA and a digital clocked ASL implementation of the TreeCompensator derived in Sec. 2.4.2 to compute error compensated output  $\hat{y}$ . The bit precisions in the estimator and the compensator blocks are primarily dictated by the number of dominant peaks in the sparse shape of the  $\eta$  distribution of the main block. The CSA and the compensator overhead amounts to 11% of the gate complexity of the MB. We assume a low error rate  $\epsilon = 10^{-4} \epsilon_{cp-avg}$  for all the gates in the CSA and TreeCompensator (marked green in Fig. 2.8(b)). We assume that the TreeCompensator computation can be pipelined since it operates only on the final outputs of the MB and the estimator. This allows the gates in the TreeCompensator to operate at a lower energy since its critical path is shorter than that of the MB.

## 2.5 Simulation Results

We demonstrate the benefits of the Shannon-inspired model of computation for a digital clocked ASL architecture of SVM classifier used for electroencephalogram (EEG) based seizure detection. The accuracy of the classifier is captured in terms of true positive (TP) rate  $p_{TP}$  and false alarm (FA) rate  $p_{FA}$ , where  $p_{TP} = \Pr\{\hat{z} = 1 | z = 1\}$  and  $p_{FA} = \Pr\{\hat{z} = 1 | z = 0\}$ , and the probabilities are estimated empirically (via leave-one-out cross-validation) [126] for the MIT-CHB EEG dataset [127] by running extensive Monte Carlo simulations. We compare the Shannon-inspired architecture (Fig. 2.8(b)) with (i) clocked ASL-based conventional serial architecture (Fig. 2.8(a)) consisting of 54,332 gates, (ii) clocked ASL-based 3-MR architecture, which replicates the conventional serial architecture thrice and takes a bitwise majority vote on their outputs, and (iii) 20 nm LV CMOS architecture, which consists of the exact same full adder-level logic network as that of the serial architecture. We compare  $p_{TP}$  vs. energy per decision and  $\epsilon_{cp-avg}$  trade-offs at a fixed decision delay of 9.7 ns and  $p_{FA} = 1\%$ .

## 2.5.1 Accuracy vs. $\epsilon_{cp-avg}$ and Energy Trade-off

We observe in Fig. 2.9(a) that Shannon-inspired architecture (Fig. 2.8(b)) can tolerate 1000× higher  $\epsilon_{cp-avg}$  compared to the conventional serial architecture (Fig. 2.8(a)) while maintaining the  $p_{TP}$  close to that of the fixed-point ideal error-free architecture. In particular, the  $p_{TP}$  for Shannon-inspired architecture is close to 93% even though  $\epsilon_{cp-avg}$  is as high as 1%. The 3-MR architecture tolerates an  $\epsilon_{cp-avg}$  up to 0.01%. It is greater than that of the serial architecture but worse by 100× when compared to the Shannon-inspired architecture. Furthermore, we show that intermediate estimator-only output ( $y_e$  in Fig. 2.8(b)) achieves lower accuracy, emphasizing the requirement to combine the two erroneous outputs ( $y_a, y_e$  in Fig. 2.8(b)) to achieve close-to-ideal accuracy.

The Shannon-inspired architecture achieves a  $3 \times$  lower energy compared to the conventional serial architecture (Fig. 2.9(b)) while maintaining  $p_{TP} =$ 93%. The 3-MR architecture, however, consumes  $2.3 \times$  more energy than the serial architecture even though it operates at a higher device error rate. This is because the energy overhead of replication offsets the energy reduction achieved by operating at higher device error rate. However, despite its high error tolerance, the Shannon-inspired architecture still requires  $1.7 \times$  more energy compared to the 20 nm LV CMOS architecture, indicating the need to explore devices with improved energy vs. error rate trade-offs and/or the use of increasingly powerful SEC techniques [128, 129, 61]. We also note in Fig. 2.9(b) that the estimator block (consisting only of the green CSA block in Fig. 2.8(b)) consumes 20% of the total energy ("Estimator Only" curve in Fig. 2.9(b)).

The reason for the effectiveness of the Shannon-inspired model in compensating for errors is the enhancement in MI  $I(Y_o; Y_a)$  due to error statistics shaping via PDB and PDR as shown Fig. 2.9(c). Despite error statistics shaping,  $y_a$  remains a poor estimate of  $y_o$ , as evident from its high classification error rate  $(1 - p_{TP})$ . Since  $I(Y_o; Y_a)$  is high, it implies that  $Y_o$  can be estimated accurately from  $Y_a$ . However, such an error compensator need not be efficient. Hence, in the Shannon-inspired model, we rely on two errorprone observations  $y_a$  and  $y_e$  to estimate  $y_o$  both efficiently and accurately. The MI  $I(Y_o; \hat{Y})$  is even higher than  $I(Y_o; Y_a)$  due to additional information about  $y_o$  contributed by  $y_e$ .

#### 2.5.2 Impact of Non-idealities and Process Variations

Next, we evaluate the tolerance of the proposed Shannon-inspired architecture to various practical non-idealities, such as, finite number of distinct clock pulse widths, process variations, and clock pulse width variations. While PDB and PDR can potentially assign a unique delay to each gate, in practice, those delays need to be further quantized to take one value out of the finite set of available distinct clock pulse widths. Figure 2.10(a) shows the  $p_{TP}$  vs.  $\epsilon_{cp-avg}$  curves for the Shannon-inspired architecture after quantizing the ideal clock pulse widths to 46 distinct pulse widths for the SVM implementation (Fig. 2.8(b)) consisting of 54,332 gates. The number of distinct clock pulse widths is of the same order as the number of gating domains explored in [116]. We observe negligible deterioration in the accuracy of Shannon-inspired architecture (in  $\epsilon_{cp-avg} < 1\%$  regime). Such gate clock pulse width quantization enables amortization of the clock pulse generation circuitry, including the sharing of the clocking transistors across different nanomagnets [116]. The clock network design is further simplified since the quantized clock pulse widths are integer multiples of the shortest reference clock, and multiple parallel dot products (in applications such as filter banks, neural networks) can share a single clock generation circuitry.

Process variations present an additional challenge in beyond-CMOS systems. We evaluate the tolerance of the Shannon-inspired approach to static within-die variations in three device parameters, namely, energy barrier  $E_b$ and damping coefficient  $\alpha$  of the nanomagnets, and clocking transistor ON current  $I_{\rm on}$ . We observe in Fig. 2.10(b) that the Shannon-inspired architecture with quantized clock pulse widths can tolerate a  $3(\frac{\sigma}{\mu})$  variations of up to 24% in each of the three device parameters. When dynamic variations in the clock pulse widths are included in addition to their quantization and process variations, we find in Fig. 2.10(c) that the Shannon-inspired architecture can tolerate a maximum deviation ( $\beta$ ) of 20% of the minimum clock pulse width  $(T_{\rm g,min})$ .

## 2.6 Discussion

In this chapter, we demonstrated how the Shannon-inspired model of computation can be employed to make digital clocked ASL implementations robust to random gate-level switching errors. Such an approach can be extended to many other spintronic devices, such as MESO [110], CoMET [111], as long as they use nanomagnet switching for information processing. Shannon-inspired techniques have previously been applied to CMOS implementations to further reduce their energy consumption via voltage overscaling [7, 129]. In contrast, ASL/spintronics provides a new way to trade off stochasticity with energy by realizing this energy-accuracy trade-off at the device level.

This chapter also showed how the hardware noise distribution in digital clocked ASL-based implementations can be shaped to enhance the robustness vs. cost trade-off discussed in Chapter 1. Such trade-off enhancements enable the use of a highly error prone but scalable physical devices (*e.g.* ASL, MESO, CoMET, MRAM, RRAM, and others) in ML inference implementations by meeting the system-level accuracy requirements despite the device-level unreliability.

# CHAPTER 3

# EFFICIENT INFERENCE VIA SPIN CHANNEL NETWORKS

## 3.1 Overview

In Chapter 2, we demonstrated how the robustness enhancement can be exploited to improve energy-efficiency of spin-based digital implementations. However, despite 1000× improvement in the robustness to gate-level switching errors, the energy-efficiency improvements fell short of outperforming CMOS implementations. Recently, Ganguly *et al.* [130] took a physicsbased approach for examining the power dissipation in spintronic switches. They identified that nanomagnetic switching consumes  $10^3 \times -to-10^4 \times$  higher switching charge ( $Q_{sw}$ ) compared to the CMOS inverter of comparable size. Such a large gap in the switching charge requirements underscores the fundamentally expensive nature of the nanomagnetic switching. ASL networks (Fig. 3.1(a)) use nanomagnetic switching at the output of every gate to implement digital logic. Hence, they require switching of a large number intermediate nanomagnets leading to a high energy consumption.

In this chapter, we propose *spin channel networks* (SCN) (Fig. 3.1(b)), where all intermediate nanomagnets are eliminated and all input nanomag-

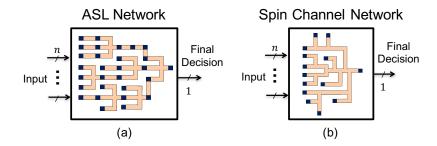


Figure 3.1: Illustration of (a) an all spin logic (ASL) network, and (b) a spin channel network (SCN) implementing an inference kernel by mapping a large n bit input vector to 1 bit decision.

nets contribute to the charge required to switch a single output nanomagnet to represent a final decision, thereby amortizing the energy consumed in switching it. It is particularly suited for inference implementations. While elimination of intermediate nanomagnetic switching is expected to enhance the energy-efficiency, it also presents following two key challenges:

(1) how does one realize arbitrary computation while accumulating analog spin currents from multiple nanomagnets?

(2) will this approach scale with the input vector dimensionality (complexity) of the inference kernel?

To address challenge (1), we show that the exponential decay property of spin current along the spin channel, a disadvantage in digital ASL networks, can be exploited to achieve energy-efficient analog dot product implementation. To circumvent the challenge (2) we employ Adaptive Boosting (AdaBoost) [131] framework to design multiple isolated tiny spin channel networks (t-SCNs) that work in unison to solve an arbitrary binary classification task. Such boosted t-SCNs achieve  $112 \times -to-22.5 \times$  and  $14 \times -to-2.5 \times$ higher energy-efficiency over conventional ASL-based and 20 nm CMOS designs, respectively, when realizing 10-to-100-dimensional binary classifiers.

## 3.2 Background

## 3.2.1 ASL Device Primitives

Figure 3.2 identifies two ASL primitives and their functionalities, that will be used to design SCNs in Sec. 3.3. The primitives are nanomagnet with a spin channel and a spin channel. In particular, nanomagnet takes input charge current  $I_c$  and injects proportional spin current  $I_{s,o}$  in the channel, where  $\beta_m$  is a proportionality constant that depends upon the device material and geometry, including the channel length  $L_c$ . The input spin current  $I_{s,in}$  into a spin channel is reduced by a factor of  $e^{\frac{L}{\lambda}}$  to generate an output spin current  $I_{s,o}$ , where L denotes channel length, and  $\lambda$  is the spin flip length [132, 133]. The layouts are obtained by following the  $\lambda$ -rules in [98].

	Conceptual diagrams	Transfer function	Symbols for schematics	Layouts
Nano- magnet with a spin channel	$I_{c}$	$= \beta_m I_c m$		Via 1 Via 2 Via Via Via Via Via Via Via Via Via Via
Spin channel	I <sub>s,in</sub> L	$I_{\rm s,o} = I_{\rm s,in} e^{-\frac{L}{\lambda}}$	$\underbrace{L}_{I_{s,in}}$	

Figure 3.2: SCN primitives derived from ASL: symbol, transfer function, and layout. Each layout grid cell is of size  $\frac{F}{2} \times \frac{F}{2} = 7.5 \text{ nm} \times 7.5 \text{ nm}$  [97].

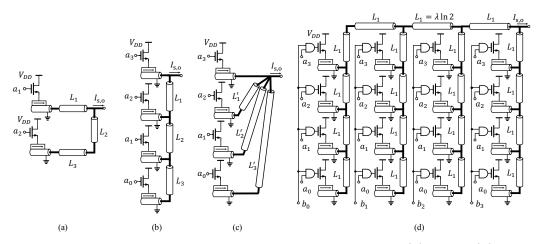


Figure 3.3: Conceptual spin channel network topologies: (a) basic, (b) *ladder*, (c) *star*, and (d) a *ladder-of-ladders* topology of a  $4 \times 4$  bit spin channel network multiplier (SCNM).

#### 3.2.2 Classifier Ensemble via Adaptive Boosting (AdaBoost)

A classifier ensemble consists of multiple weak classifiers. Each weak classifier is computationally simple but inaccurate, i.e., with  $p_e$  close to 0.5. However, decisions of the weak classifiers can be combined to obtain a highly accurate final decision. Adaptive boosting (AdaBoost) [134] is a technique to train these weak classifiers sequentially. Each weak classifier is specifically trained to correct errors made by the other weak classifiers trained earlier (see [134] for the training algorithm). Let the output label of *i*th weak classifier be denoted as  $\hat{y}_i = f_{\mathbf{w}_i}(\mathbf{x})$ , where  $f_{\mathbf{w}_i}(.)$  denotes the *i*th weak classifier function parametrized by weight vector  $\mathbf{w}_i$ , which is computed during training. The final decision  $\hat{y}_f$  is computed by linearly combining the weak classifier decisions  $\hat{y}_i$ , followed by thresholding as shown below:

$$\sum_{i=1}^{M} \alpha_i \hat{y}_i \underset{\hat{y}_f=-1}{\overset{\hat{y}_f=-1}{\gtrless}} 0 \tag{3.1}$$

where output weights  $\alpha_i$ s of the linear combiner are also learned during the training phase.

## 3.3 Spin Channel Networks

#### 3.3.1 Basic Concept

Spin channel networks exploit the exponential decay of spin current along spin channels for efficient computation. They compute via weighted analog accumulation of spin currents by careful choice of spin channel lengths. SCNs are composed of the two primitives defined in Fig. 3.2. The most basic SCN consists of two nanomagnets connected using spin channels having different lengths is shown in Fig. 3.3(a). The resulting output spin current  $I_{s,o}$ , is approximately given by

$$I_{\rm s,o} = \beta_m I_c \left( a_1 m_1 e^{\frac{-L_1}{\lambda}} + a_2 m_2 e^{\frac{-(L_2 + L_3)}{\lambda}} \right)$$
(3.2)

where  $a_1, a_2 \in \{0, 1\}$  are the digital Boolean inputs,  $m_1, m_2 \in \{-1, 1\}$  denote the directions of magnetization vectors of two nanomagnets,  $\lambda$  denotes spin

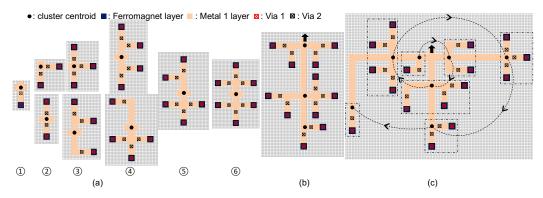


Figure 3.4: Layouts of spin channel networks: (a) a set of clusters along with their centroids, (b) a ladder topology of three clusters, (c) a star-of-ladders layout topology of a  $4 \times 4$  bit spin channel network multiplier. The layout grid cell is of size  $7.5 \text{ nm} \times 7.5 \text{ nm}$  [97].

flip length,  $I_c$  denotes the ON current of the NMOS transistors. Each bit  $a_i$  controls the charge current through one nanomagnet, and the corresponding spin current is weighed by a factor exponential in its channel length. More complex SCNs can be designed to achieve weighted accumulation of spin currents from M nanomagnets placed at lengths  $L_i$ s, where  $i \in \{0, \ldots, M-1\}$ .

#### 3.3.2 SCN Topologies

For M > 2, multiple circuit topologies can achieve the same input-to-output transfer function (up to a scaling constant) depending upon how the nanomagnets and spin channels are interconnected. For example, conceptual diagrams of two extreme topologies, namely the *ladder* and the *star* topology, are shown for M = 4 in Fig. 3.3(b) and 3.3(c), respectively. In the ladder topology, all nanomagnets share a single spin channel that connects them to the output node. The star topology, on the other hand, consists of a unique channel connecting each nanomagnet to the output node.

Weighted accumulation of spin currents in SCNs can be used to efficiently implement multiplication in analog. The  $M \times N$  bit SCN multiplier (SCNM) in Fig. 3.3(d) takes two charge-domain digital operands A and B having bitwidths M and N, respectively, and generates an output spin current proportional to their product  $A \times B$ . The SCNM consists of  $M \times N$  input nanomagnets, each contributing spin current corresponding to a partial product. Individual partial products are computed by the AND gates with bits  $a_i$ s and  $b_j$ s of operands A and B, respectively. The AND gates drive the gate of the NMOS, thereby controlling the input charge current through the SCN nanomagnets. Figure 3.3(d) shows a *ladder-of-ladders* topology of a 4 × 4 bit SCNM, where four vertical ladders are connected horizontally in a ladder topology. Note that at all the spin channel lengths are multiples of  $\lambda \ln 2$  in order to achieve appropriate weighing (in the powers of two) of spin currents corresponding to individual partial products. The output spin current of this  $4 \times 4$  bit SCNM is given by:

$$I_{\rm s,o} = I_{\rm s,lsb} \sum_{i,j=0}^{3} a_i b_j 2^{(i+j)}$$
(3.3)

where the unit spin current  $I_{s,lsb}$  corresponds to the least significant partial product for a given NMOS ON current  $I_c$ . For the SCNM in Fig. 3.3(d),  $I_{s,lsb} = \frac{\beta_m I_c}{27}$ . The signs of these operands can be accounted for by changing the magnetization vector directions of the corresponding magnets (for A), and by using a differential supply [135] (for B). The energy consumption of such an SCNM is given by:

$$E_{\text{mult}}(A,B) = \left[ I_c^2 T_g(R_{\text{spin}} + R_{\text{mos}}) + C_g V_g^2 + E_{\text{and}} \right] \sum_{i,j=0}^3 a_i b_j$$
(3.4)

where  $R_{\rm spin}$  denotes the series resistance of the nanomagnet and channel,  $E_{\rm and}$  denotes the energy consumed in switching of the AND gate, while  $R_{\rm mos}$ and  $C_g$  denote the ON resistance and gate capacitance of the transistor, respectively. The gate voltage  $V_g$  is applied to switch ON the NMOS for  $T_g$ duration.

The other topologies such as *star-of-ladders*, *star-of-stars* and *ladder-of-stars* are also possible, and this topological degree of freedom will be explored while identifying the energy-efficient layout in the following Sec. 3.3.3.

## 3.3.3 Hierarchical Layout Construction

Topological schematic diagrams in Fig. 3.3 are idealized and convey the SCN functionality at a very high level. They neither account for spin current branching at the spin channel junction, nor the physical constraints of component placements and maintaining spin channel lengths. We address both

of these issues by developing precise layouts for SCN circuits, and obtain input-to-output transfer function from SCN layouts.

For layouts, we choose F = 15 nm, where F denotes the DRAM half pitch [97, 98]. All SCN layouts need to satisfy  $\lambda$ -rules described (in terms of F) in [97]. For example, the layout pitch between any two contacts needs to be at least 4F. The value of channel length  $L_c$  turns out to be 5F (in Fig. 3.2) as a direct consequence of  $\lambda$ -rule constraints. Similarly,  $\lambda$ -rules impose constraints on minimum distance between nanomagnet and a spin channel, and two parallel spin channels.

The layouts of more complex SCNs are particularly challenging, since there is a trade-off between satisfying  $\lambda$ -rule constraints and the magnitude of the output spin current, and hence the energy consumption. We propose a hierarchical construction of SCN layouts. We define nine primitive topologies referred to as *clusters* (see Fig. 3.4(a)). Each nanomagnet in a cluster contributes identical spin currents to the output node, referred to as the *cluster centroid*. The layouts of the clusters are fixed per  $\lambda$ -rules. These clusters can be connected in various topologies, such as a ladder, star or a ladder-of-stars, to generate layouts of more complex SCNs in a hierarchical manner. An illustrative ladder topology of three clusters is shown in Fig. 3.4(b). Once the clusters are connected, only the lengths between their centroids need to be adjusted to achieve appropriate weighing of the corresponding spin currents and simultaneously satisfy  $\lambda$ -rules.

Figure 3.4(c) shows a star-of-ladders layout topology of a 4×4 bit SCNM. Each cluster centroid  $J_k$  generates spin current corresponding to  $p_k$ , where  $p_k$  is defined as the sum of partial products having identical binary weight k as follows:

$$p_k = \left(\sum_{\substack{i,j=0\\i+j=k}}^3 a_i b_j\right) \tag{3.5}$$

The final output spin current in Eq. (3.3) can be computed as:

$$I_{\rm s,o} = I_{\rm s,lsb} \sum_{k=0}^{6} 2^k p_k \tag{3.6}$$

where the binary weighing of  $2^k$  among the spin current contributions is achieved by adjusting spin channel lengths between them. The clusters are sequentially placed in a spiral order along three ladders as shown in Fig.

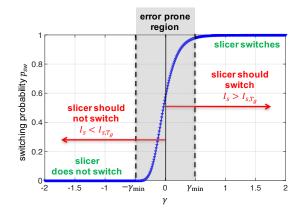


Figure 3.5: The switching probability  $p_{\rm sw}$  of stochastic slicer as a function of  $\gamma$  (given approximately by Eq. (3.7)) with  $E_b = 35kT$  and T = 300 K. The dotted outlines of the gray region mark the minimum energy operating point (MEOP) at  $\gamma_{\rm min} = 0.5$  corresponding to slicer switching accuracy of 99%.

3.4(c). Thus, along a single ladder, the minimum channel length between any two consecutive clusters corresponds to the spin current weighing of  $2^{-3}$ , thus allowing sufficient spacing to satisfy  $\lambda$ -rules. Hence, this layout topology effectively spreads out the nanomagnets radially. The actual channel lengths in the layout are chosen via extensive simulations using SPICE-based circuit models of spin current injection and propagation in spin devices [133] in order to account for spin current branching at the spin channel junctions.

## 3.3.4 Stochastic Slicer

The output of SCN circuits is an analog spin current. We use a nanomagnet as the final decision device. It acts as a sink for the spin current and thresholds it to produce the final decision represented by its magnetization vector. The nanomagnetic switching is stochastic due dominant thermal noise in the nanomagnet [102]. Hence, we refer to such decision generating nanomagnet as a *stochastic slicer*. The stochastic slicer switches when the magnetization direction of the corresponding nanomagnets flips due to the input spin current.

In this work, we operate the stochastic slicer for the fixed duration of  $T_g$ . If the slicer switches during this duration, it corresponds to final decision  $\hat{y} = 1$ , otherwise,  $\hat{y} = -1$ . The slicer is reset after every decision. For a given duration  $T_g$ , the probability that slicer switches  $p_{sw}$  can be approximated as a function of its input spin current  $I_s$  as follows:

$$p_{\rm sw}(I_s) \approx \left(\frac{1}{2}\right)^{\left[\left(\frac{\beta_1}{\ln 2}\right)^{-\gamma}\right]}$$
(3.7)

where  $\gamma = \left(\frac{I_s - I_{s,T_g}}{I_{s,T_g}}\right)$ ,  $I_{s,T_g}$  denotes the spin current for which  $p_{sw}(I_{s,T_g}) \approx 0.5$ , and  $\beta_1 = \frac{\pi^2 E_b}{4kT}$ . In particular,  $E_b$  denotes the energy barrier of the nanomagnet, while k and T denote Boltzmann constant and absolute temperature, respectively. The spin current value  $I_{s,T_g}$  is a device-dependent constant for a given switching duration  $T_g$ . Equation (3.7) provides a good approximation of the  $p_{sw}$  expression in [102], when  $|I_s|, I_{s,T_g} \gg I_{crit}$ , where  $I_{crit}$  denotes the minimum spin current required for nanomagnet to switch with probability 1 as  $T_g \to \infty$ . The stochastic slicer strives realize the thresholding operation:

$$I_s \underset{\hat{y}=-1}{\overset{\hat{y}=1}{\geq}} I_{s,T_g} \tag{3.8}$$

However, due its stochastic nature, stochastic slicer probabilistically makes switching errors, i.e., it switches with certain non-zero probability even when  $I_s < I_{s,T_g}$ , and vice versa as shown in Fig. 3.5. Thus, there exists a trade-off between input spin current magnitude  $I_s$  (proportional to energy consumption) and switching probability  $p_{sw}$  (see Eq. (3.7)). There exists a minimum energy operating point (MEOP) for a target switching probability. For example, as shown in Fig. 3.5,  $|I_s| > 1.5I_{s,T_g}$  to achieve slicer switching accuracy of 99%. This MEOP of slicer dictates the minimum charge current  $I_{c,\min}$ through each nanomagnet required for SCN-based binary classifier to achieve certain classification error probability  $p_e$ . For a fixed decision delay and error probability, MEOP for SCN-based binary classifiers is uniquely defined by the value of  $I_{c,\min}$  as shown in Sec. 3.5.

#### 3.3.5 CMOS Driver

Figure 3.6(a) and (b) show the abstract model and transistor-level schematic of the CMOS input driver in a 14 nm technology, respectively. The nanomagnet is controlled by an NMOS, which should switch ON only when  $a_i = 1$ and  $b_j = 1$ . This is achieved via a CMOS NOR gate driving the gate of

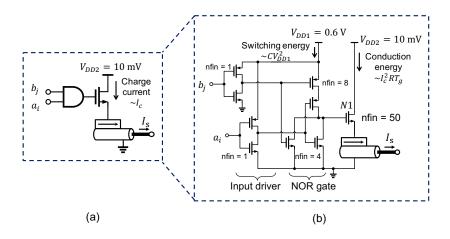


Figure 3.6: Detailed schematic of the CMOS input driver designed and simulated using 14 nm HP FinFET ASU predictive technology models [136], where nfin denotes number of fins in the FinFET [137].

the NMOS N1 as shown in Fig. 3.6(b). The inputs to the NOR gate are driven by identical inverters, who receive ideal step inputs. The NMOS N1is sized to provide a charge current of  $I_{c,\min}$ , while satisfying  $V_{DS} < 10 \text{ mV}$ . The gate voltage of N1 gets raised to 600 mV, turning it ON in the linear region with overdrive voltage  $\geq 350 \text{ mV}$ . The NOR gate is sized so that the CMOS driver switches within 50 ps while driving N1 and the inverters are minimum sized. We simulate this schematic using 14 nm HP FinFET ASU predictive technology models [136] to estimate its switching delay and energy consumption.

## 3.4 Design of SCN-based Classifiers

#### 3.4.1 Linear Support Vector Machine (SVM) Classifier

A linear SVM classifier can be realized using the proposed SCNs by connecting multiple SCNMs in parallel (Fig. 3.7(c)) and one stochastic slicer to generate the final classification decision. The SCNM and slicer symbols are defined in Fig. 3.7(a) and (b), respectively. Each multiplier generates spin current corresponding to the product  $x_iw_i$ , where  $x_i$  and  $w_i$  denote *i*th dimension of input feature vector  $\mathbf{x}$  and the weight vector  $\mathbf{w}$ , respectively. Both  $x_i$  and  $w_i$  are fixed-point binary numbers. The spin currents at the output of SCNMs accumulate in a common channel (generating  $I_{s,o}$ ) and feed into the stochastic slicer.

The inputs are kept ON for the duration  $T_g$  since the stochastic slicer requires that much time to produce its decision. Noting that the stochastic slicer thresholding involves a comparison of its input spin current with  $I_{s,T_g}$ , Eq. (2.1) of SVM can be realized as follows:

$$I_{s,o} = (\mathbf{w}^T \mathbf{x} + b) I_{s,lsb} + I_{s,bias} \underset{\hat{y}=-1}{\overset{\hat{y}=1}{\geq}} I_{s,T_g}$$
(3.9)

where  $I_{s,bias}$  is additional bias current.

When  $I_{s,o} = I_{s,T_g}$ , the slicer switches with probability 0.5. In SVM, this operating point would occur when the input feature vector  $\mathbf{x}$  lies on the classifier hyperplane, i.e. when  $\mathbf{w}^T \mathbf{x} + b = 0$ , resulting in

$$(I_{s,o})|_{\mathbf{w}^T\mathbf{x}+b=0} = I_{s,T_g} = I_{s,bias}$$
 (3.10)

To avoid large bias currents, we modify the feature vector  $\mathbf{x}$  to  $(\mathbf{x} + d\mathbf{1})$ , where  $\mathbf{1}$  denotes the all-one vector and d is a constant, thereby transforming Eq. (3.9) into

$$\left[\mathbf{w}^{T}(\mathbf{x}+d\mathbf{1})\right]I_{\mathrm{s,lsb}} + I_{\mathrm{s,bias}} \underset{\hat{y}=-1}{\overset{\hat{y}=1}{\geq}} I_{s,T_{g}}$$
(3.11)

where  $I_{s,bias}$  is now given by:

$$I_{\rm s,bias} = I_{s,T_g} + bI_{\rm s,lsb} - \left[d\left(\sum_i w_i\right)\right]I_{\rm s,lsb}$$
(3.12)

The bias current  $I_{s,bias}$  is generated by having an additional magnet with a supply current  $I_{c,bias}$  as shown in Fig. 3.7. If the signed precision of  $x_i$  is M bits, we choose d to be  $2^{M-1}$ . This makes  $(x_i + d)$  an unsigned number, removing the need for differential supply. The sign of  $\mathbf{w}$  is accounted for by changing the magnetization vector directions of the corresponding spincurrent injecting nanomagnets appropriately.

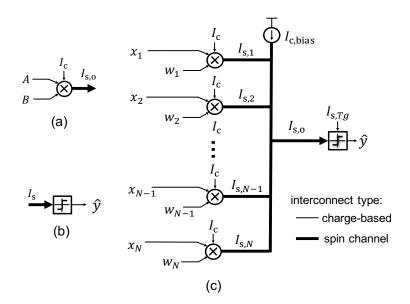


Figure 3.7: SCN-based linear SVM classifier: (a) SCNM symbol, (b) stochastic slicer symbol, and (c) the SCN-based *N*-dimensional linear SVM classifier architecture.

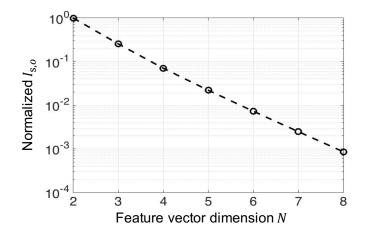


Figure 3.8: Normalized magnitude of  $I_{s,o}$  in Fig. 3.7(c) as a function of classifier dimensionality N, when  $x_1 = c_1 \neq 0$ ,  $w_1 = c_2 \neq 0$ , and  $x_i = w_i = 0$  for all  $i \in \{2, \ldots, N\}$ , where  $c_1$  and  $c_2$  are some constants.

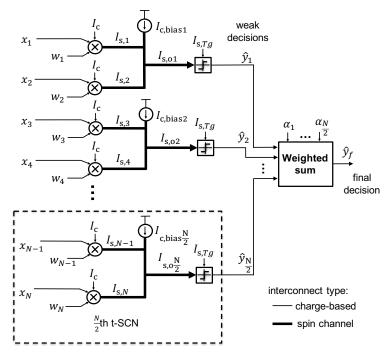


Figure 3.9: Boosted tiny SCNs architecture: adaptively boosted ensemble of  $\frac{N}{2}$  tiny SCNs (t-SCNs), where each t-SCN consists of 2 SCNMs in parallel and one stochastic slicer to implement a two-dimensional linear SVM classifier.

## 3.4.2 Classifier Dimensionality Scaling via Boosted Tiny SCNs (t-SCNs)

The exponential decay of spin current exploited in SCNM makes it very hard to route the output spin current to another block as doing so inevitably incurs a significant loss in the spin current magnitude. This severely limits the ability to scale the classifier dimensionality (Fig. 3.7(c)), which requires the N multiplier output spin currents to be routed to a single stochastic slicer. Assuming a circular layout of N SCNMs with the slicer at its center, we estimate the loss in the spin current  $I_{s,o}$  magnitude as a function of classifier dimensionality N as shown in Fig. 3.8, when  $x_1 = c_1 \neq 0$ ,  $w_1 = c_2 \neq 0$ , and  $x_i = w_i = 0$  for all  $i \in \{2, \ldots, N\}$ , where  $c_1$  and  $c_2$  are some constants. Recall from Fig. 3.5 that the stochastic slicer requires minimum magnitude of the input spin current in order to operate accurately for a given switching delay. Thus, the exponential loss in the output spin current magnitude results in exponentially increasing classifier energy consumption in order to maintain classifier accuracy.

In order to address the problem, we limit the dimensionality of the SCNbased linear SVM classifier to only two dimensions, and refer to the resulting

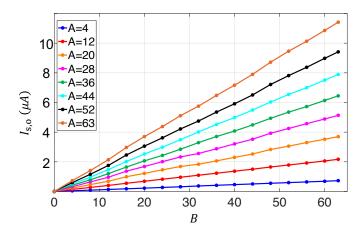


Figure 3.10: Simulated transfer function of a  $6 \times 6$  bit SCNM realizing  $A \times B$ .

design as tiny SCN (t-SCN). We then employ AdaBoost to design an ensemble of multiple such t-SCNs to implement an arbitrary N-dimensional binary classification task. Figure 3.9 shows the boosted t-SCNs architecture. In particular, given an N-dimensional input feature vector, each t-SCN observes only two unique feature dimensions and computes its local decision  $\hat{y}_i$ . These local decisions could be inaccurate with higher probability, and hence are referred to as weak decisions. The final weighted sum block combines these weak decisions to obtain the final decision  $\hat{y}_f$  as per Eq. (3.1). We restrict the number of weak classifiers to  $\frac{N}{2}$  so that computational complexity of the boosted t-SCNs architecture is similar to the standard N-dimensional linear SVM implementation (Fig. 3.7(c)).

It is important to note that, in the boosted architecture, the output spin current of the channel network gets processed locally, and only the binary weak decisions are routed to the final weighted sum block, thus requiring much shorter spin interconnect routing within each t-SCN. It is straightforward to convert the binary slicer decisions  $\hat{y}_i$   $i \in \{1, \ldots, \frac{N}{2}\}$  to equivalent voltage [138] and then route it using charge interconnects. We designed the linear combiner in Eq. (3.1) in conventional digital 14 nm CMOS. Its complexity in terms of the full adder count is less than 5% of the total complexity of the  $\frac{N}{2}$  t-SCNs. One can also employ other schemes, such as Boolean logic, Winner-Take-All, to efficiently combine binary t-SCN outputs, achieving similar energy benefits.

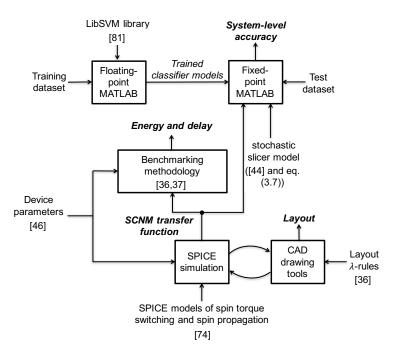


Figure 3.11: Simulation methodology.

## 3.5 Simulation Results

## 3.5.1 Simulation Methodology

Our simulation methodology is shown in Fig. 3.11. We first develop the SCNM layout and obtain its transfer function via SPICE-based simulations. We employ SPICE-based spin device models [133] for SCNM schematic simulations and use CAD drawing tools for corresponding layouts and  $\lambda$ -rule checks. We assume the material parameters provided in Table 3.1. We design and characterize a 6×6 bit SCNM using the SPICE-based spin device models [133]. The channel lengths between the clusters in the SCNM layouts are repeatedly adjusted until the appropriate spin current weighting is achieved and all  $\lambda$ -rules are satisfied. Figure 3.10 shows the interpolated SCNM transfer function after carrying out detailed spice simulations for 289 different A and B values. The observed  $\left(\frac{\sigma}{\mu}\right)$  of the deviations from linearity in output spin current is 2%. Such good linearity was achieved, in part, because process variations (such as line edge roughness etc.) and length quantization between two clusters are not yet accounted for. However, this does indicate that there is still room to budget such impact of more variations within in-

Table 3.1: The SPICE-based models developed in [133] were employed to simulate all spin channel network designs in this chapter. Hence, most of the device and material parameters of nanomangets and the copper channel are chosen as mentioned in [133]. Few device parameters that were specifically chosen for the designs in this chapter are given in this table.

Variable	Value	
Saturation magnetization [103]	$250 \times 10^3 \mathrm{A/m}$	
Effective internal anisotropic field [103]	$16 \times 10^4 \mathrm{A/m}$	
Damping coefficient[103]	0.007	
Nanomagnet dimensions	$30\mathrm{nm}\times30\mathrm{nm}\times10\mathrm{nm}$	
Slicer dimensions	$30\mathrm{nm}\times80\mathrm{nm}\times2\mathrm{nm}$	
Spin channel width	30 nm	
Spin channel thickness	100 nm	

herent tolerance of machine learning classifiers. Also, since all such variations within the layouts are static, one can employ retraining [64] to alleviate their impact. We leave this exploration for future work.

We use the SCNM transfer function and stochastic slicer model (based on Eq. (3.7) in the main text and analysis in [102]) for system-level classifier accuracy predictions. We estimate the classification error probability  $p_e$  of boosted t-SCN implementation in MATLAB. We use LIBSVM [139] to train all SVM classifiers. The classifier training always happens in floating-point precision in MATLAB. We then quantize the trained classifier model and test data to have 6 bit precision, which we found out to be sufficient. We carry out 6-fold cross validation over the available data to estimate the average accuracy of all classifier implementations.

We employ benchmarking methodology [97, 98] to estimate energy and delay for all classifier implementations compared in Sec. 3.5.2. For ASL implementations, we assume ASL device consisting of nanomagnets with improved anisotropy reported in [103]. We also assume ASL gates to be clocked using a MOSFET to avoid static power consumptions. For digital CMOS implementations, we use the delay vs. energy curve of 20 nm LV CMOS FO4 inverter reported in [103], and assume activity factor of 33%.

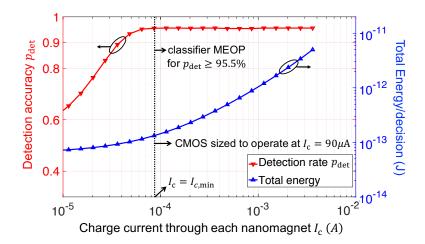


Figure 3.12: Accuracy  $p_{det}$  and total energy vs.  $I_c$  trade-off for 10dimensional boosted t-SCN classifier operating at a final decision delay of 3 ns. The minimum energy operating point (MEOP) is achieved at  $I_{c,\min} = 90 \,\mu\text{A}$ .

## 3.5.2 Results

We demonstrate the effectiveness of proposed approach for two classification tasks: (1) 10-dimensional (10D) breast cancer detection (UCI repository dataset[140], [141]), and (2) 100-dimensional (100D) face detection (MIT CBCL dataset [142]). We quantify classification accuracy in terms of detection rate  $p_{det} = 1 - p_e$ , where  $p_e$  is classification error probability.

For each t-SCN, there exists a trade-off between NMOS current  $I_c$  and weak decision delay  $T_g$  for fixed  $p_{sw}$ . We choose  $T_g = 2.5$  ns throughout this chapter to make sure that CMOS driver switching energy  $\leq \approx 33\%$  of the total energy. We compare the energy consumption and accuracy of 10D and 100D classifier implementations at a fixed final decision delay of 3 ns and 4 ns, respectively. The remaining duration accounts for the delay of CMOS driver switching, slicer reset, and weighted logic block operation. In particular, the CMOS driver can be switched within 50 ps. For 10D classifier, weighted logic block operation can be approximated as a majority operation. We choose identical  $I_c$  for all weak classifiers. Given  $I_c$  and  $T_g$ ,  $I_{c,bias}$  is chosen according to Eq. (3.12) for each weak classifier.

For a fixed final decision delay (of 3 ns), the trade-off between the accuracy and total energy consumption of the 10D boosted t-SCN classifier is shown in Fig. 3.12 as a function of  $I_c$ . As expected, both accuracy and total energy decrease with  $I_c$ . The accuracy degradation occurs due to reductions in  $p_{sw}s$ 

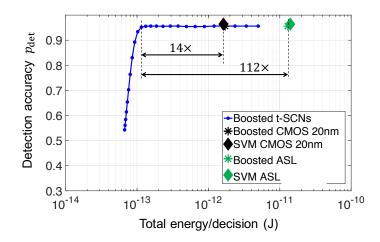


Figure 3.13: Classification accuracy  $p_{det}$  vs. energy trade-off for different 10-dimensional classifier implementations operating at a final decision delay of 3 ns.

of the stochastic slicers. For accuracy of 95.5%, the classifier MEOP (defined in Sec. 3.3.4) is achieved at  $I_{c,\min} = 90 \,\mu\text{A}$ . Hence, we size the NMOS N1 to provide  $I_c$  of 90  $\mu\text{A}$  at  $V_{DD2} = 10 \,\text{mV}$  (see Fig. 3.6).

Figure 3.13 shows the accuracy vs. energy trade-off for different 10D classifier implementations. Boosted t-SCN classifier achieves at least  $112 \times$  lower energy per decision compared to that of the conventional boosted ASL implementation, while maintaining accuracy. Such large energy savings can be attributed to the elimination of all intermediate switching nanomagnets in the spin channel network implementation. It also achieves  $14 \times$  lower energy compared to boosted 20 nm LV CMOS digital implementation, while operating at the identical final decision delay. We also observe that both boosted CMOS and boosted ASL implementations achieve energy consumption similar to the corresponding N-dimensional linear SVM implementations. For 100D classifier (Fig. 3.14), the energy benefits of boosted t-SCN implementation reduce to  $2.5 \times$  and  $22.5 \times$  over CMOS and ASL SVM implementations, respectively. This is primarily because of higher  $I_c$  requirements for its weak SVM classifiers, resulting from lower class separability of the dataset. Thus, the energy benefits are a function of the input data statistics as well. We only compare dynamic energy here, but leakage energy will be the least for SCN implementations due to having fewer transistors compared to both CMOS and ASL implementations. For all ASL implementations, we assume that

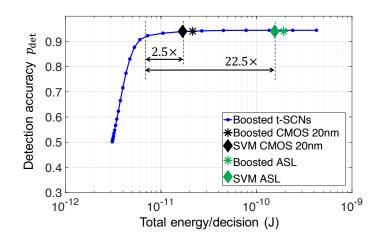


Figure 3.14: Classification accuracy  $p_{det}$  vs. energy trade-off for different 100-dimensional classifier implementations operating at a final decision delay of 4 ns.

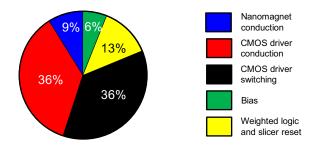


Figure 3.15: Category-wise energy breakdown for 10-dimensional boosted t-SCN implementation operating at a decision delay of 3 ns and with  $I_{c,\min} = 90 \,\mu\text{A}$ .

the clocking transistors are shared across multiple nanomagnets [116], significantly amortizing their energy consumption.

In Fig. 3.15, we observe that the CMOS driver conduction energy and switching energy are comparable, and together dominate the energy consumption of the 10D boosted SCN classifier. The CMOS driver is expensive to switch due to large size of NMOS N1, which is necessary due to large charge current requirements ( $I_{c,\min} \approx 100 \,\mu\text{A}$ ) of SCN classifiers. Conduction energies of CMOS driver and nanomagnet add up to a constant  $V_{DD2}I_{c,\min}T_g$ . These trends are similar to ASL implementations.

## 3.6 Related Works

There have been few works that exploit the stochastic nature of nanomagnetic switching to achieve energy-efficiency. In [138], the switching behavior of magnets in super-paramagnetic regime was shown to resemble the dynamics of a Boltzmann machine, and thus a nanomagnetic network was trained to implement certain inference tasks. Stochastic magnets were also employed for energy-efficient random number generation [143] in stochastic computing, as well as for spike generation [144] in spiking neural network implementation.

Several research efforts have explored the neuromorphic design space using spin-devices in order to achieve energy-efficiency. While it is clear that the switching of nanomagnet naturally implements thresholding function of a neuron, these approaches use additional devices (resistive memory [145], domain wall magnets [146, 147, 148]) to achieve synaptic weighing of spin currents feeding into the nanomagnet. In [135], multiple binary weighted CMOS drivers along with a clever nanomagnet configuration were employed to obtain spin-current weighing in cellular neural network implementation.

There exists work at the architectural-level to fully exploit the advantages of emerging spin-device configurations such as racetrack memory [149, 150]. For example, high area-efficiency and serial access of racetrack memory was exploited to achieve reconfigurable precision [151] and efficient logic operations [152]. In [153], a novel data converter design was proposed by exploiting the serial structure of racetrack memory devices.

## 3.7 Discussion

In this chapter, we proposed spin channel networks where multiple input nanomagnets contribute to the spin current required to switch a single decision nanomagnet. These networks exploit diffusive spin current for efficient local computation to achieve very high energy-efficiency and ensemble of such isolated networks can solve any given classification task. Note that the exponential decay of the spin current and error-prone switching of the nanomagnets have been viewed as disadvantages in the digital all spin logic (ASL) implementations. This chapter demonstrates how the synergistic application of algorithmic techniques can turn apparent disadvantages of the emerging devices into their strengths, and improve the system-level robustness vs. cost trade-off significantly in the process. For example, the proposed spin channel networks with their diffusive spin currents and unreliable nanomagnets achieved comparable accuracy at  $10 \times$  lower energy per decision compared to conventional digital 20 nm CMOS implementations.

# CHAPTER 4

# EFFICIENT INFERENCE VIA MRAM-BASED DEEP IN-MEMORY ARCHITECTURE

## 4.1 Overview

As discussed in Sec. 1.2, the energy to access data from memory is significantly more than doing computations on that data [18]. This challenge is particularly exacerbated for deep net implementations, since they require millions of stored parameters to be accessed and computed upon. To address this concern, recently, *in-memory computing* (IMC) approach has emerged as one of the promising directions. In this approach, one strives to embed processing elements (PEs) in the periphery of memory bit-cell array (BCA) [108, 65], in order to directly readout a computed function of the stored data as a part of the memory access. For example, given a matrix  $\mathbf{W}$  stored in BCA and input activation vector  $\mathbf{x}$ , IMC strives to compute output vector  $\mathbf{a} = \mathbf{W}\mathbf{x}$ within a single BCA read access. This is particularly useful for efficient deep net inference implementations, since almost all memory intensive operations can be viewed as matrix vector multiplications (MVMs). However, they often need to be implemented in analog, due to severe area/pitch-matching constraints arising due to the high density nature of BCAs. Such embedded in-memory computation, while fast and energy-efficient, remains vulnerable to process variations [65]. Thus, in this approach, one needs to balance the energy/latency cost vs. robustness in order to maximize the benefits.

Much of the recent work in IMC focuses on SRAMs [108, 65, 154, 155, 156, 157, 158, 159, 160, 161, 162, 163, 164, 165, 156]. Most notably, [108, 65] proposed an SRAM-based deep in-memory architecture (DIMA) to achieve a multi-bit vector dot product without requiring any modification in the standard 6T SRAM BCA. However, for state-of-the-art deep nets, on-chip SRAM may not be sufficient to store all the parameters, requiring highly energy and latency expensive off-chip DRAM accesses. Hence, it is necessary to ex-

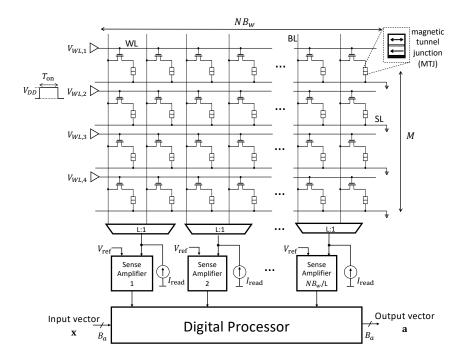


Figure 4.1: Digital implementation of matrix-vector multiplication (MVM) with weights stored in 1T-1MTJ MRAM bitcell array (BCA).

tend such an approach to emerging memories such as RRAM/PCM/MRAM, which are attractive alternatives due to their higher density and non-volatility. They seem particularly promising for hardware implementations of deep nets due to the possibility of having larger on-chip memory and potentially fewer memory write operations.

In the area of emerging memories, most of the recent works have focused on RRAM/PCM-based in-memory computing [166, 167, 168, 169, 170, 171, 8, 172] by exploiting multi-bit storage and/or large ON/OFF resistance ratio in RRAM/PCM bitcells. However, MRAM-based in-memory computing remains unexplored today even though MRAM has started to become available as a standard digital memory as a part of commercial process development kits [173]. This is, in part, because low ON/OFF resistance ratio and the absence of multi-state bitcells in MRAM makes it challenging to achieve multi-bit dot product computation within MRAM BCA.

In this chapter, we explore MRAM-based DIMA (MRAM-DIMA) to achieve multi-bit matrix-vector multiplication (MVM) within a single read operation. We employ a standard MRAM-BCA, without any modifications, to preserve its density. We propose modified peripheral circuits to achieve such multibit computation, even though, unlike RRAM/PCM, each bitcell stores only 1 bit. The proposed MRAM-DIMA is shown to potentially achieve  $10 \times$  and  $20 \times$  lower energy and delay, respectively, compared to digital MRAM implementations with the matrix stored in an identical MRAM array.

## 4.2 Preliminaries

## 4.2.1 Notation

In this chapter, we assume the following MVM computation needs to be executed:

$$\mathbf{a} = \mathbf{W}\mathbf{x} \tag{4.1}$$

where **W** denotes a  $M \times N$  weight matrix, and **x** and **a** denote the input and output vectors, respectively. Each weight is denoted as  $w_{ij} \forall i \in$  $\{1, \ldots, M\}, j \in \{1, \ldots, N\}$  and is quantized to  $B_w$  bits. Each element of the vectors **x** and **a**, denoted as  $x_i$  and  $a_i$ , respectively, is quantized to  $B_a$  bits.

#### 4.2.2 Digital MRAM Architecture

Figure 4.1 shows a diagram of a conventional digital MRAM architecture with a BCA of size  $N_{\rm row} \times N_{\rm col} = M \times NB_w$  which stores the weight matrix **W**. This BCA has sourcelines (SLs) and wordlines (WLs) perpendicular to the bitlines (BLs). The MRAM bitcell consists of an NMOS access transistor and a magnetic tunnel junction (MTJ) (1T-1MTJ). The stored bit is read by sensing the resistance across the MTJ, which is low when the two magnetization vectors within the MTJ are parallel to each other (P state), and is high when they are antiparallel (AP state). In this chapter, bit value 1 (0) corresponds to the P (AP) state. We denote total bitcell resistance (conductance) by  $R_{\rm P}$  ( $G_{\rm P}$ ) when MTJ is the P state, and by  $R_{\rm AP}$  ( $G_{\rm AP}$ ) when it is in the AP state.

During an MRAM read operation, a constant current of  $I_{\text{read}}$  is passed through the bitcells and the voltage developed on the BLs is sensed to determine the MTJ state. Access transistors of a single row are activated for

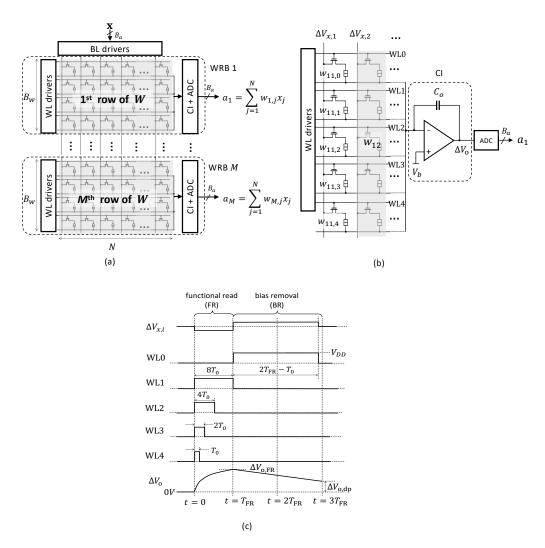


Figure 4.2: The proposed MRAM-based deep in-memory architecture (MRAM-DIMA): (a) the overall architecture consisting of M parallel word-row blocks (WRBs) each sharing N BL drivers but possessing separate WL drivers and current integrators (CIs), (b) schematic of a WRB, and (c) the read access timing diagram.

 $T_{\rm on}$  duration by driving its WL to  $V_{DD}$ . A sense amplifier (SA) converts the voltage on the BL to a bit decision. Since the SA is typically wider than a bitcell, it is shared across multiple columns via L: 1 multiplexers (typical value of L is from 8 to 32). Thus,  $\frac{N}{L}$  weights are read per cycle, which are then input to a digital processor consisting of  $\frac{N}{L}$  parallel multipliers followed by an adder tree for computing a dot product.

The average energy required by the MRAM-digital architecture to implement the MVM computation in Eq. (4.1) is given by:

$$E_{\text{digital}} = MNB_w \left[ (I_{\text{read}} V_{DD} T_{\text{on}} + E_{SA}) + LC_{\text{wl-cell}} V_{DD}^2 \right] + E_{\text{proc}}$$
(4.2)

where  $E_{SA}$  and  $E_{\text{proc}}$  denote the sense amplifier energy and the total digital processor energy, respectively,  $T_{\text{on}}$  is the ON time of a single row during BCA read operation, and  $C_{\text{wl-cell}}$  denotes the WL capacitance per bitcell. Similarly, the delay to complete the MVM operation is given by:

$$T_{\text{digital}} = MLT_{\text{on}} + T_{\text{proc}} \tag{4.3}$$

where  $T_{\text{proc}}$  denotes the delay of the digital processor in Fig. 4.1.

## 4.3 Voltage Driven MRAM-based Deep In-memory Architecture

#### 4.3.1 Overall Architecture

Figure 4.2(a) shows the MRAM-DIMA consists of a conventional MRAM BCA and peripheral blocks including WL drivers, BL drivers, and current integrators (CIs) on SLs. The weights are stored in the BCA in a column major format. A set of  $B_w$  rows of BCA storing one row of **W** constitutes a word-row block (WRB). Each WRB implements a vector dot product to compute one element of the output vector **a** in Eq. (4.1). All WRBs operate in parallel, sharing the BLs, but possess separate WL drivers and SL CIs, as shown in Fig. 4.2(a).

#### 4.3.2 MRAM-DIMA Operation

In this subsection, we describe the operation of a single WRB in detail. Figure 4.2(b) shows a schematic of WRB 1 with  $B_w = 5$  and  $B_a = 4$ . Analog voltages  $\Delta V_{x,i} = -x_i V_{\rm lsb}$  are applied to the BLs via per column DACs ( $V_{\rm lsb}$ denotes the DAC output resolution). Next, a functional read (FR) [108] step is initiated, in which the bottom  $B_w - 1$  WLs are activated simultaneously by applying pulse-width modulated (PWM) access pulses with the *b*th WL turned ON for a duration of  $2^{B_w-b-1}T_0$ . Total FR phase duration is  $T_{FR} = 2^{B_w-2}T_0$ . All SL currents are summed in the CI, followed by an ADC to generate the digital outputs  $a_i$ s. At the end of the FR phase, the resulting CI output voltage  $\Delta V_{o,FR}$  is given by:

$$\Delta V_{\text{o,FR}} = \underbrace{\frac{T_{\text{o}} V_{\text{lsb}}}{C_{o}} \left[ \Delta G \sum_{j=1}^{N} w_{ij} x_{j} \right]}_{=\Delta V_{\text{o,dp}} \text{ (dot product)}} + \underbrace{\frac{T_{\text{o}} V_{\text{lsb}}}{C_{o}} \left[ (2^{B_{w}-1}-1) \sum_{j=1}^{N} G_{j} x_{j} \right]}_{=\Delta V_{\text{o,bias}} \text{ (bias)}}$$
(4.4)

where  $\Delta G = G_P - G_{AP}$ ,  $G_P = \frac{1}{R_P + R_{\text{mos}}}$ ,  $G_{AP} = \frac{1}{R_{AP} + R_{\text{mos}}}$ ,  $C_o$  denotes the capacitor in the CI, and  $G_j = G_{AP}$  (=  $G_P$ ), when  $w_{ij} \ge 0$  ( $w_{ij} < 0$ ).

The bias term  $\Delta V_{o,bias}$  in Eq. (4.4) is generated by the non-zero bitcell current when storing a zero-valued bit. This bias term is removed in bias removal (BR) step which discharges  $C_o$  appropriately by enabling only WL0 as shown in Fig. 4.2(c).

#### 4.3.3 Scaled-up MRAM-DIMA

The current integrator in Fig. 4.2(b) has a finite output swing, i.e.,  $\Delta V_{\rm o} \leq \Delta V_{\rm o,max}$ . This output swing limitation is exploited to naturally implement clipped ReLU activation function in DNNs. However, in order to achieve correct computation, it is also necessary that  $\Delta V_{\rm o,FR} < \Delta V_{\rm o,max}$ . This condition is challenging to meet as vector dimension N increases (see Eq. (4.4)), since  $\Delta V_{\rm o,bias}$  gets canceled only during the BR step.

To design MRAM-DIMA for large N, we propose swing budgeting via *phase multiplexed computation* within each WRB. We partition the  $MB_w \times N$  BCA vertically into three sub-arrays, each consisting of  $\frac{N}{3}$  columns as shown in Fig. 4.3(a). All three sub-arrays share their SLs and CIs. However, we

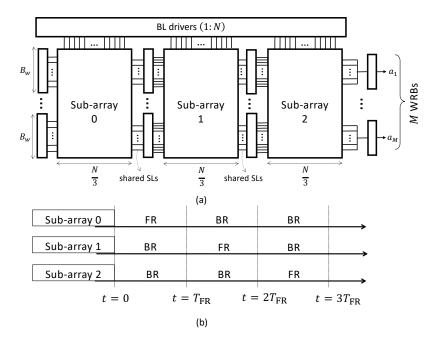


Figure 4.3: Realizing scaled-up MRAM-DIMA via phase multiplexed computations: (a) architecture, and (b) timing diagram.

introduce distinct WL drivers for individual sub-arrays to operate two in the BR phase, while the third in FR step as shown in Fig. 4.3(b). This approach drastically reduces the resulting  $\Delta V_{o,FR}$  at  $t = T_{FR}$  and  $t = 2T_{FR}$ . For LeNet-300-100 DNN on MNIST dataset, it is estimated that  $Pr{\Delta V_{o,FR} > \Delta V_{o,max}} < 0.01\%$  at any given time instant using the behavioral models developed in Sec. 4.3.4, resulting in negligible system-level accuracy drop.

## 4.3.4 Modeling Energy, Delay, and MTJ Process Variations

The average energy consumption of MRAM-DIMA for implementing  $M\times N$  MVM is:

$$E_{\text{dima}} = MNB_w \left[ \left( \frac{2^{B_w} - 2}{B_w} \right) \bar{x} V_{\text{lsb}} G_{\text{cell}} V_{DD} T_0 + C_{\text{wl-cell}} V_{DD}^2 \right] + ME_{\text{adc}} + ME_{\text{CI}} + NE_{\text{dac}}$$
(4.5)

where  $G_{\text{cell}} = \frac{G_P + G_{AP}}{2}$ ,  $\bar{x}$  denotes the average value of **x** across its N components,  $E_{\text{adc}}$ ,  $E_{\text{CI}}$ , and  $E_{\text{dac}}$  denote the energy consumed in the ADC, CI, and DAC, respectively. The DAC is assumed to be equipped with an opampbased voltage follower to drive the required current in BL [174]. Hence,  $E_{\text{dac}}$ 

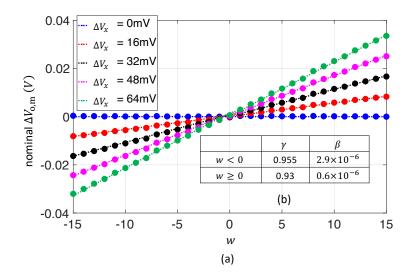


Figure 4.4: Transfer function of a scalar multiplier in WRB operating on two operands w and  $x \propto \Delta V_x$  for  $B_w = 5$  in a commercial 22 nm CMOS-MRAM process: (a) simulated (circles) and modeled (lines) and (b) estimated fitting parameters.

and  $E_{\rm CI}$  are dominated by the respective opamp bias current energies. The value of  $\bar{x}$  is estimated from the distribution of activations in DNNs. Compared to the digital MRAM energy consumption (in Eq. (4.2)), the BCA energy is reduced due to two factors: (1) the small value of  $\bar{x}$ , thanks to high activation sparsity in DNNs, caused by ReLU, and (2) the reduced read current per bitcell, since multiple bitcell currents are aggregated before ADC operation. The MRAM-DIMA delay is:

$$T_{\rm dima} = 3 \times 2^{B_w - 2} T_0 + T_{\rm adc} + T_{\rm dac} \tag{4.6}$$

where  $T_{\text{adc}}$ ,  $T_{\text{dac}}$  denotes the delay of ADC and the DAC, respectively. The delay of MRAM-DIMA is nearly constant, resulting in the speed-up increasing with the number of rows M.

The expression in Eq. (4.4) for an analog vector dot product output  $\Delta V_{o,dp}$ ignores the impact of circuit non-idealities such as the body-effect of access transistors, virtual ground voltage bounce in the current integrator, and process variations in the MTJ. To account for these non-idealities, we model the analog output of a multiplier with a  $B_w$  bit operand w and an analog input x in the WRB as follows:

$$\Delta V_{\text{o,m}} = \frac{\gamma T_0 V_{\text{lsb}}}{C_o} \left[ \Delta G w + (2^{B_w - 1} - 1)\beta \right] x + \eta \tag{4.7}$$

where  $\eta$  denotes the spatial noise arising due to process variations in the MTJ. The fitting parameters  $\beta$  and  $\gamma$  are obtained via circuit simulations in a commercial 22 nm CMOS-MRAM process (see in Fig. 4.4).

Process variations at the multiplier output arises from  $R_{\rm MTJ}$  variations across different bitcells. Hence,  $\eta$  in Eq. (4.7) follows a zero-mean Gaussian distribution with the variance  $\sigma_{\eta}^2$  given by:

$$\sigma_{\eta}^{2} = \left(\frac{T_{0}V_{\rm lsb}}{C_{0}}x\right)^{2} \left((2^{B_{w}-1}-1)^{2}G_{0}^{2} + \sum_{b=1}^{B_{w}-1}4^{b-1}G_{b}^{2}\right) \left(\frac{\sigma}{\mu}\right)_{\rm G-bc}^{2}$$
(4.8)

where  $\left(\frac{\sigma}{\mu}\right)_{\text{G-bc}}$  denotes  $\sigma$ -to- $\mu$  ratio of  $G_{\text{cell}}$ , and  $G_b \in \{G_P, G_{AP}\}$  is the conductance of the *b*-th cell in a column, where  $b \in \{0, \ldots, B_w - 1\}$ . It is to be noted that both nominal  $\Delta V_{\text{o,m}}$  and  $\sigma_{\eta}$  scale linearly with  $T_0$  and  $V_{\text{lsb}}$  keeping the signal-to-noise ratio unchanged. We estimate  $\left(\frac{\sigma}{\mu}\right)_{\text{G-bc}} \leq 6\%$  via Monte Carlo circuit simulations for both P and AP states.

## 4.4 Simulation Results for Voltage Driven MRAM-DIMA

#### 4.4.1 Design Choices

In this chapter, we choose  $B_w = 5$  and  $B_a = 4$ , since these are typical DNN precision requirements for inference [175, 176]. For MRAM-DIMA, we assume a single-slope ADC architecture with a shared ramp generation circuit across the WRBs, typically employed in column-parallel ADCs in CMOS image sensors [177]. From the measured 8 bit, 7.1 MS/s 65 nm single-slope ADC results in [108], we conservatively estimate  $E_{\rm adc} = 0.84 \, \rm pJ/sample$  and  $T_{\rm adc} + T_{\rm dac} = 25 \,\rm ns$  for 4 bit precision, with  $T_{\rm adc}$  dominating the delay. Furthermore, assuming  $\Delta V_{\rm o,max} = \frac{V_{DD}}{3} = 300 \,\rm mV$  and  $V_{\rm lsb} = 4 \,\rm mV$ , the systemlevel output swing requirements for mapping DNN computations described in Sec. 4.4.3 dictate  $T_0 = 1 \,\rm ns$ , which is previously shown to be achievable

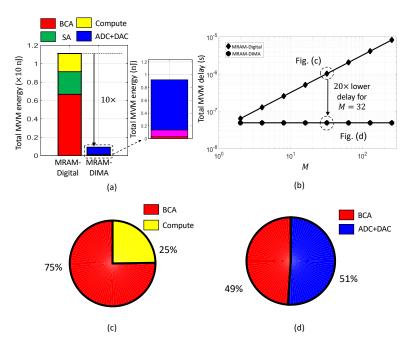


Figure 4.5: Simulation results for MVM operation: (a) energy breakdown, (b) delay as a function of M, (c) component-wise delay contributions in digital MRAM, (d) component-wise delay contributions in MRAM-DIMA for  $M \times N = 32 \times 192$ ,  $B_w = 5$ , and  $B_a = 4$ .

[178, 179, 180] without significant energy overhead.

In the digital MRAM architecture, we use a BCA of identical capacity as in DIMA to store the weight matrix. Using the measured offset voltage variation in SA [181] and observed  $\left(\frac{\sigma}{\mu}\right)_{\text{G-bc}}$  in the PDK, we derive  $I_{\text{read}} = 40 \,\mu\text{A}$ . Similarly, we assume  $T_{\text{on}} = 3 \,\text{ns}$ , L = 8, and  $E_{SA} = 40 \,\text{fJ}$  from the results reported in [182, 181, 183, 184]. We obtain energy and delay of a full-adder via SPICE simulations and use it to estimate  $E_{\text{proc}}$  and  $T_{\text{proc}}$  in Eq. (4.2) and Eq. (4.3), respectively.

#### 4.4.2 Energy and Throughput Benefits

In Fig. 4.5, we plot energy and delay models in Sec. 4.3.4 for  $M \times N = 32 \times 192$ MVM computation corresponding to a  $3 \times 3$  convolutional layer with 64 input and output channels. The MRAM-DIMA achieves a  $10 \times$  lower energy as shown in Fig. 4.5(a), primarily due to elimination of SA, and BCA energy reduction due to activation sparsity and reduced read current per bitcell. In MRAM-DIMA,  $E_{dac}$  and  $E_{CI}$  dominate due to their large opmap bias

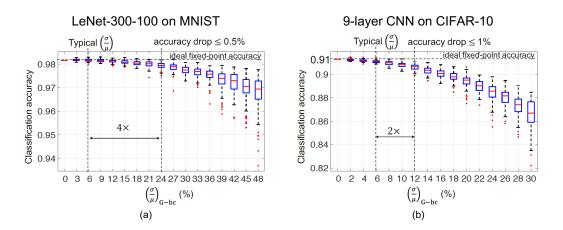


Figure 4.6: Classification accuracy vs  $\left(\frac{\sigma}{\mu}\right)_{\text{G-bc}}$  obtained using the models in Sec. 4.3.4 for: (a) LeNet-300-100 on MNIST dataset, and (b) nine-layer CNN on CIFAR-10 dataset. For each value of  $\left(\frac{\sigma}{\mu}\right)_{\text{G-bc}}$ , 100 instances of the BCA were generated to obtain the statistics shown in the error bars.

currents. The aggregated maximum current on SLs in a WRB is smaller than maximum current on single BL due to activation sparsity and phase multiplexed bias removal. This leads to smaller bias current energy in  $E_{\rm CI}$ than that of the voltage follower in  $E_{\rm dac}$ . The delay benefits of MRAM-DIMA over digital MRAM scale linearly with M due to the inherent parallelism in MRAM-DIMA (see Fig. 4.5(b)). A delay reduction of up to 20× is achieved for M = 32. While memory access delay dominates in digital MRAM, (see Fig. 4.5(c)),  $T_{\rm adc}$  constitutes  $\approx 50\%$  of  $T_{\rm dima}$  (Fig. 4.5(d)).

#### 4.4.3 Impact on System-level Accuracy

We estimate the impact of MRAM-DIMA's analog computation and process variations on the system-level accuracy of: (a) LeNet-300-100 MLP on MNIST dataset, and (b) a nine-layer CNN on CIFAR-10 dataset. These networks were trained using DoReFa-net [185] training methodology. The activations pass through a clipped ReLU activation function and are quantized between [0, 1]. The weights lie in the range between  $[-\alpha_{ij}, +\alpha_{ij}]$ , where  $\alpha_{ij}$  is the scaling parameter for the *i*-th filter in the *j*-th layer, and they all lie in the range [0.03, 0.15]. Limitation on the voltage swing at the CI output ( $\Delta V_{o,max}$ ) naturally implements clipped ReLU, while the  $\alpha_{ij}$ s are implemented by appropriately scaling  $T_0$ . We emulate MRAM-DIMA's computation in PyTorch by using nominal behavioral models described in Sec. 4.3.4 for deterministic

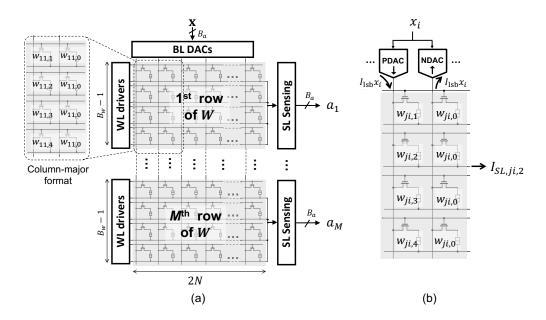


Figure 4.7: Current driven MRAM-DIMA design details: (a) column-major storage pattern of matrix **W** in BCA, and (b) current DAC driven BLs.

non-idealities and add bitwise random noise in bitcell conductance values to model MTJ variations. The MRAM-DIMA is able to tolerate a  $4\times$  and  $2\times$  higher  $\left(\frac{\sigma}{\mu}\right)_{\text{G-bc}}$  than its typical value while maintaining < 1% drop in accuracy compared to ideal fixed-point computation for LeNet-300-100 and nine-layer CNN as shown in Fig. 4.6(a) and (b), respectively.

## 4.5 Current Driven MRAM-DIMA

In this section, we present a modified version of, where the BLs are driven by current DACs instead of the voltage DACs as discussed in Sec. 4.3. We discuss corresponding changes in the BCA storage and BL/SL peripherals.

#### 4.5.1 BCA

As shown in Fig. 4.7(a), the weight matrix **W** is stored in the BCA in a column-major format, where  $w_{ij,0} \ldots w_{ij,4}$  denote individual bits of 5 bit scalar elements  $w_{ij}$  represented in 1's complement format. Here  $w_{ij,0}$  indicates the sign, while  $w_{ij,1}$  is treated as the most significant bit (MSB). Thus, 5 bit scalar element  $w_{ij}$  occupies eight bitcells, with the sign bit  $w_{ij,0}$  replicated

four times in the adjacent column. Again, a WRB is formed by combining four rows of BCA storing one row of  $\mathbf{W}$  (similar to voltage driven design in Fig. 4.2(a). Furthermore, the pulse-width modulation for the WLs in each WRB is also same as that of the voltage driven design. However, the BR phase in voltage driven design is eliminated here by employing both P-type and N-type current DACs as discussed in the next subsection.

#### 4.5.2 Current Driven BLs

BLs are driven by current DACs as shown in Fig. 4.7(b). This enables independent control of the current through each BL. Assuming ideal SL sensing, *i.e.* with zero input impedance looking into SL sensing circuit, the current through bitcell storing  $w_{ji,2}$  can be written as:

$$I_{ji,2} = \frac{1}{G_{\rm BL,i}} \left[ G_{\rm AP} + \Delta G w_{ji,2} \right] x_i I_{\rm lsb}$$

$$\tag{4.9}$$

where  $G_{\text{BL},i}$  denotes total conductance looking into the BL in *i*th column storing  $w_{ji,2}$ ,  $\Delta G = G_{\text{P}} - G_{\text{AP}}$ , and  $I_{\text{lsb}}$  denotes the LSB current of BL DAC. Eq. (4.9) indicates that for achieving accurate dot product in single WRB, each BL needs to have identical conductance looking into it, *i.e.*,  $G_{\text{BL},i} = G_{\text{BL}}$  $\forall i$ . Note that this does not occur naturally, since each  $G_{\text{BL},i}$  is a function of data stored in that column. For now, we assume that BL conductances for all columns are equal to  $G_{\text{BL}}$ , and address BL conductance equalization in Sec. 4.5.4.

It is also to be noted that only the second term in the above Eq. (4.9) is desired for the dot product computation, while the first term is an additional bias that needs to be cancelled. It arises due to the fact that  $G_{AP} \neq 0$ , *i.e.*, the bitcell contributes a non-zero current despite storing a 0 bit. This bias can be removed via an N-type current DAC connected to BL in the adjoining column storing replicated sign bit  $w_{ji,0}$ , as shown in Fig. 4.7(b). The resulting *unbiased* SL current contribution from bitcell pair  $w_{ji,2}$  and  $w_{ji,0}$  is given as:

$$I_{\rm SL, ji, 2} = \frac{\Delta G}{G_{\rm BL}} (w_{ji, 2} - w_{ji, 0}) x_i I_{\rm lsb}$$
(4.10)

It is to be noted that such bias removal via N-type current DAC also naturally enables signed computation, where resulting SL is negative for

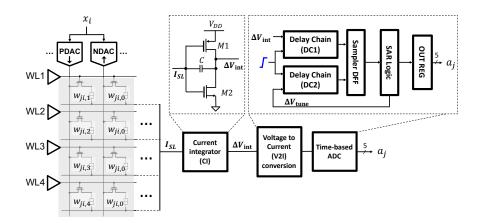


Figure 4.8: Block diagram of SL sensing circuits in MRAM-DIMA.

negative valued weight  $w_{ji}$ .

#### 4.5.3 Time-based Sensing Circuits

Figure 4.8 shows a block diagram of sensing circuits at the SL. They consist of three stages, namely, current integrator (CI), voltage-to-time conversion, and time-based ADC. In CI, the incoming SL currents are integrated to develop voltage change  $\Delta V_{\text{int}}$  at its output, which is proportional to the dot product computed in that WRB. The CI is implemented as an inverter with a capacitive feedback. This integrator output  $\Delta V_{\text{int}}$  is converted to a digital code via a time-based ADC to produce 5 bit output  $a_j$ . In particular, it is applied to a delay chain DC1 (see Fig. 4.8) such that it proportionately changes the time it takes for the rising edge to pass through the delay chain. Thus, it is compared with a tunable reference voltage  $\Delta V_{\text{tune}}$  by comparing the delays of DC1 and DC2 via sampling D-flipflop. Then,  $\Delta V_{\text{tune}}$  is adjusted via standard successive approximation register and a current DAC, until  $\Delta V_{\text{tune}}$  is within one LSB of  $\Delta V_{\text{int}}$ . The resulting digital code corresponding to  $\Delta V_{\text{int}}$  is read out via the OUT register.

The simplicity and component-efficiency of the proposed sensing circuit offers one crucial advantage – it enables pitch-matched layouts of this block, *i.e.*, the layout of each sensing circuit is pitch-matched to the height of each WRB, which equals to four times individual bitcell pitch. We discuss these layout constraints further in Sec. 4.6.

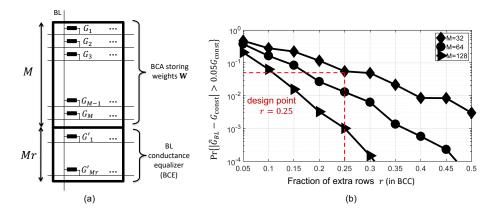


Figure 4.9: BL conductance equalization in MRAM-DIMA: (a) pictorial representation of BL conductance equalizer (BCE) with respect to BCA, and (b) probability of encountering a column whose conductance can not be compensated within  $\pm 5\%$  of  $G_{\text{const}}$  as a function of the fraction of extra rows r constituting BCE.

#### 4.5.4 BL Conductance Variations across Columns

When BLs are driven by current DACs, each bitcell current contribution gets scaled by total BL conductance in that column  $G_{\text{BL},i}$  Eq. (4.9), deteriorating the accuracy of the dot product in each WRB. Hence, we propose to explicitly equalize the BL conductance across the by adding extra rows in the BCA, as shown in Fig. 4.9. We refer to these additional rows as *BL conductance* equalizer (*BCE*), and denote the equalized conductance by  $\hat{G}_{\text{BL}}$ . Note that the BCE array is an identical BCA receiving the same WL pulses. The data the BCE is written such that total BL conductance (include BCA and BCE) is equal for each column. For example, in Fig. 4.9, the values of bitcell conductances  $G'_1, \ldots, G'_{Mr}$  in BCE depend upon the bitcell conductances  $G_1, \ldots, G_M$  in BCA column above them.

Let r denote a ratio of the numbers of rows in BCE to the number of rows in BCA. It is clear that the BL conductances are equalized perfectly across the columns, if r = 1. However, it also involves 100% overhead for such equalization. Hence, a key question is, can one achieve good enough equalization across the columns at much smaller values of r? This question can be answered rigorously as follows.

Let  $G_{\text{const}}$  denote the desired conductance after equalization for each column. One way to quantify the quality of equalization is by determining the probability  $(p_{\text{bad}})$  of encountering a column whose conductance cannot be

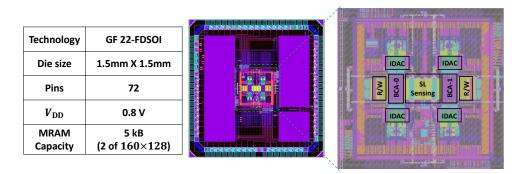


Figure 4.10: Layout and details of the chip prototype of current driven MRAM-DIMA.

compensated within 10% of  $G_{\text{const}}$ , despite having r rows in BCE, *i.e.*,

$$p_{\text{bad}} = \Pr\left\{ \left| \hat{G}_{\text{BL}} - G_{\text{const}} \right| > 0.05 G_{\text{const}} \right\}$$
(4.11)

Under the assumption of Bernoulli data distribution, the relationship  $p_{\text{bad}}$ vs. r can be derived analytically for different values of BCA rows M, as shown in Fig. 4.9(b). Observe that, for a given  $p_{\text{bad}}$ , the required overhead rdecreases, as expected, with increasing M. In this design, we have M = 32, and choose r = 0.25 to achieve  $p_{\text{bad}} \approx 5\%$  as a sweet-spot between the accuracy vs. overhead of the conductance equalization.

## 4.6 Prototype Chip Tape-out

Figure 4.10 shows the details and the layout of our chip prototype of current driven MRAM-DIMA designed using GF-22 nm-FDSOI process development kit. The prototype was designed to implement  $32 \times 128$  MVM computation. It consists of  $160 \times 256$  MRAM crossbar which is columnwise split into two sub-arrays, namely BCA-0 and BCA-1. The SL sensing is located in the middle, reducing the impact of SL parasitic resistance. The standard digital MRAM read & write circuits (R/W blocks in Fig. 4.10) are located on the other sides of the two subarrays. Note that the SL sensing circuits are pitchmatched to the BCAs. This was challenging due to the high density of MRAMs. In particular, MRAM bitcells are so small that the pitch width of a *single* D flip-flop in the OUT Register (see Fig. 4.8) is comparable to the pitch of *eight* MRAM bitcells. Such tight pitch-matching constraints were met by

component-efficient time-based sensing circuits (see Sec. 4.5.3) and extensive manual layout efforts. At the end, SL sensing circuits corresponding to the consecutive WRBs are stacked sideways, and are pitch-matched to eight BCA rows.

During the prototype chip measurements, we found it challenging to achieve desired dot product accuracy due to significantly low signal-to-noise ratio (SNR) at the SL currents. In particular, we observed that the SL sensing circuits are overwhelmed by the sum of mismatch currents in all current DAC pairs, resulting in a large bias current at the SL that is not proportional to the dot product. This observation, while perplexing at first, motivated the detailed SNR analysis for IMC employing resistive crossbars presented in Chapter 5, where we analytically provide the reason behind this observation.

## 4.7 Discussion

In this chapter, we proposed an MRAM-based deep in-memory architecture to achieve matrix-vector multiplication within a single BCA read operation. It employed standard MRAM-BCA, without any modifications and modified peripheral circuits to achieve multi-bit computation even though each MRAM bitcell stores only 1 bit. We discussed both voltage-driven and current driven MRAM-DIMA. We showed that MRAM-DIMA can potentially achieve  $10 \times$  and  $20 \times$  lower energy and delay, respectively, compared to digital MRAM implementation with the matrix stored in an identical MRAM array.

While we designed a prototype chip implementing current-driven MRAM-DIMA, we observed prohibitively low SNR during measurements. This observation motivated a detailed SNR analysis of resistive crossbar IMCs presented in Chapter 5. Based on our analysis, we concluded that such amplification of the impact of input DAC mismatches does occur in current driven MRAM crossbars due to lower resistance states in MRAM and the larger input impedance ( $\approx 2 \,\mathrm{k}\Omega$ ) of sensing circuits. This being a fundamental device-level limitation, we recommend on-chip learning based error compensation techniques to improve the output SNR.

## CHAPTER 5

# SNR ANALYSIS OF IN-MEMORY COMPUTING EMPLOYING RESISTIVE CROSSBAR ARRAYS

### 5.1 Overview

In this chapter, we focus on understanding the compute SNR of IMCs employing emerging 1T-1R crossbar memories, such as RRAM, PCM, or MRAM. This SNR is significantly affected by the presence of process variations in the bitcell resistance as well as in the input digital-to-analog converters (DACs). While the recent works [166, 167, 168, 169, 170, 171, 8, 172] present individual design points, the understanding of how SNR is affected by the nonzero impedance of peripheral sensing circuits, the dimensionality of the dot product, and inherent ON/OFF resistance ratio for the bitcells remains incomplete. It is important to complete this understanding in order to identify the true capabilities of resistive IMC crossbars and develop design guidelines. This indeed is the goal of this chapter.

We conduct an SNR analysis of voltage driven resistive crossbars employed for in-memory MVM computations. The analysis is device-agnostic and is applicable to any resistive crossbar. Specifically, we use our analysis to comprehend the SNR trade-offs for PCM, MRAM, as well as RRAM. We consider the impact of device and circuit non-idealites such as the finite impedance  $R_s$  of the sensing circuits, and the mismatch in bitcells and input DACs. We also capture the impact of output analog-to-digital converters (ADCs) via accounting for clipping and quantization noise. In the case of voltage DAC (VDAC) driven crossbars, we show how SNR trades-off with  $R_s$ , the number of columns N in the crossbar, as well as the normalized energy per operation. Finally, we also consider current DAC driven crossbars and identify their challenges compared to voltage driven crossbars.

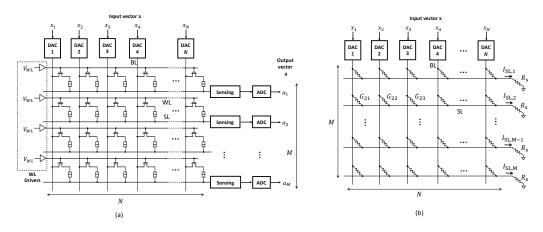


Figure 5.1: IMC with resistive non-volatile memories: (a) block diagram of a typical architecture, and (b) its abstraction for steady-state SNR analysis presented in this chapter.

## 5.2 Analysis Setup

Figure 5.1(a) shows a block diagram of a canonical crossbar architecture employing an  $M \times N$  resistive BCA and associated peripherals required for MVM computation. Each bitcell in the BCA consists of an access transistor controlled via horizontal wordlines (WL) and a resistive memory device whose conductance is proportional to the value of data stored in the bitcell. Note that MRAM devices can have only two distinct conductance states, storing single bit per bitcell. However, RRAM/PCM devices can potentially store multiple bits within single bitcell via multiple distinct conductance states.

The input vector  $\mathbf{x}$  is applied to the bitlines (BLs) via current/voltage DACs (see Fig. 5.1(a)), while the BCA stores the matrix  $\mathbf{W}$  with every element stored in the corresponding bitcell. The elements of the output vector  $\mathbf{a} = \mathbf{W}\mathbf{x}$  needs to be estimated from the row-wise horizontal sourceline (SL) currents via SL sensing circuits followed by analog-to-digital converters (ADCs). Notice that during the IMC operation all WLs are turned ON. Hence, for the SNR analysis in this chapter, we make the following two abstractions (see Fig. 5.1(b)): (i) the BCA is represented as a purely resistive crossbar array, where each resistance value includes both the ON resistance of the access transistor and the resistance of the memory device, and (ii) we represent the SL sensing circuits as a resistance  $R_s$ , which represents the Thevenin equivalent resistance looking into the input of the SL sensing circuit. We account for the impact of the ADC, such as quantization noise and clipping noise, by referring those to the input of the SL sensing circuits.

We denote the resistance of a bitcell in *i*th row and *j*th column as  $R_{ij} \in \{R_{\text{on}}, R_{\text{off}}\}$  and corresponding conductance  $G_{ij} = \frac{1}{R_{ij}} \in \{G_{\text{on}}, G_{\text{off}}\}$ , where we assume each bitcell stores a single bit with two distinct resistance states: low resistance state  $R_{\text{on}}$  and high resistance state  $R_{\text{off}}$ , and  $G_{\text{on}}$  and  $G_{\text{off}}$  denote the corresponding conductance values. We consider following three memory device technologies since they cover a wide range of  $R_{\text{off}}/R_{\text{on}}$  values:

- MRAM [186]:  $R_{\rm on} \approx 3 \,\mathrm{k}\Omega$  and  $R_{\rm off} \approx 6 \,\mathrm{k}\Omega \implies \frac{R_{\rm off}}{R_{\rm on}} \approx 2$
- RRAM [172]:  $R_{\rm on} \approx 25 \,\mathrm{k\Omega}$  and  $R_{\rm off} \approx 300 \,\mathrm{k\Omega} \implies \frac{R_{\rm off}}{R_{\rm on}} \approx 12$
- PCM [8]:  $R_{\rm on} \approx 40 \, \mathrm{k}\Omega$  and  $R_{\rm off} \approx 3 \, \mathrm{M}\Omega \implies \frac{R_{\rm off}}{R_{\rm on}} \approx 75$

where we define the ratio  $R_{\text{off}}/R_{\text{on}}$  as the *resistive contrast* of the bitcell.

In a resistive crossbar, the signal of interest is the SL current  $I_{\text{SL},i}$  in *i*th row whose *ideal* value is given by  $\left(\sum_{k=1}^{N} V_k G_{i,k}\right)$  where  $V_k$  denotes the voltage applied at the *k*th BL. We first derive an exact closed-form expression of the nominal value of  $I_{\text{SL},i}$  as a function of  $R_s$  and N. Then, we consider the impact of variations and noise on  $I_{\text{SL},i}$ . Specifically, the noise SL current  $I_{\text{SL}}$ for each row is written as (dropping index *i* for simplicity):

$$I_{\rm SL} = I_{\rm sig} + I_{\rm bm} + I_{\rm dm} + I_{\rm cn} + I_{\rm qn}$$
 (5.1)

where  $I_{\text{sig}}$  denotes nominal SL current in the absence of any variations, and the other terms are defined below:

- $I_{\rm bm}$ : impact of mismatch variations in the bitcell conductances. It is well known that  $R_{\rm on}$  and  $R_{\rm off}$  values in the crossbars vary by 4%-to-6%.
- $I_{\rm dm}$ : impact of mismatch variations in the input DACs. Due to the threshold voltage variations, the ON current of each DAC finger varies by ~ 5%, resulting in an input-dependent variation in the total current of each current DAC. This phenomenon is also present in voltage DACs if they are constructed by cascading a current DAC and a transimpedance amplifier (TIA). Such voltage DAC design is preferred due to the strict area constraints in the IMC crossbars.
- $I_{\rm cn}$ : noise originated due to the clipping of  $I_{\rm SL}$  beyond range  $-I_{\rm clip}$  to  $I_{\rm clip}$ . When SL sensing circuits convert  $I_{\rm SL}$  to voltage before ADC,

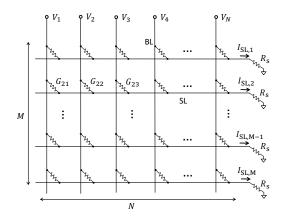


Figure 5.2: Voltage driven crossbar.

 $I_{\rm clip}$  is dictated by the input voltage swing of the ADC. For advanced process technologies such as CMOS 22 nm,  $I_{\rm clip}$  is typically in the range of a few  $\mu$ A for time-based successive approximation ADCs discussed in Sec. 4.5.3.

•  $I_{qn}$ : noise due to quantization via ADC.  $I_{qn} \sim U\left(-\frac{I_{clip}}{2^{B_{adc}}}, \frac{I_{clip}}{2^{B_{adc}}}\right)$ , where U(L, H) denotes the uniform noise distribution between L and H, and  $B_{adc}$  denotes the ADC bit precision. For most resistive IMC crossbars with on-chip ADCs, B is typically chosen as 3-to-6 bits.

We define the SNR at the SL as:

$$SNR = \frac{\mathbb{E}[I_{sig}^2]}{\mathbb{E}[I_{bm}^2] + \mathbb{E}[I_{dm}^2] + \mathbb{E}[I_{cn}^2] + \mathbb{E}[I_{qn}^2]}$$
(5.2)

where the expectation  $\mathbb{E}[\cdot]$  is taken over the joint distribution of data, variations, and noise.

## 5.3 Voltage Driven Crossbars

In a voltage driven crossbar (see Fig. 5.2), the *j*th BL is driven by voltage  $V_{\text{BL},j} = V_{\text{DC}} + V_j$ , where  $V_{\text{DC}}$  denotes the DC bias voltage,  $V_j = x_j V_{\text{lsb}}$  inputdependent small-signal voltage,  $x_j$  denotes *j*th element of vector **x**, and  $V_{\text{lsb}}$ denotes minimum DAC voltage step corresponding to least significant bit (LSB). For  $I_{\text{SL},i}$  to be proportional to the  $V_j$ s, the DC voltage of SLs needs to be set to  $V_{\text{DC}}$ . This can be achieved by using the same TIA topology as that of the voltage DACs in the SL sensing to convert resulting  $I_{\rm SL,i}$  into voltage. For the analysis in this chapter, we assume this to be the case. Furthermore, in practice,  $V_{\rm DC}$  of each DAC and each SL also varies due to the threshold voltage variations in the constituent transistors of the corresponding TIAs. In this chapter, however, we do not include those variations in our analysis since they are input-independent and can be compensated by a one-time tuning of the TIAs per chip.

Notice that even if *i*th crossbar row stores a vector of all zeros, *i.e.*, when  $G_{ij} = G_{\text{off}}$ ,  $j \in 1, ..., N$ , the resulting current  $I_{\text{SL},i} \neq 0$  since  $G_{\text{off}} \neq 0$ . In order to remove any non-zero DC current when  $G_{ij} = G_{\text{off}}$ , it is a common practice to [171, 8]: (i) apply differential inputs on the BLs, *i.e.*,  $V_{2k-1} = -V_{2k}$ ,  $k \in \{1, ..., \frac{N}{2}\}$ , and (ii) employ two bitcells  $(G_{i,2k-1}, G_{i,2k})$  to store a single signed bit as follows:

$$b_{i,2k} = \begin{cases} 1 & \text{if } G_{i,2k-1} > G_{i,2k} \\ 0 & \text{if } G_{i,2k-1} = G_{i,2k} \\ -1 & \text{if } G_{i,2k-1} < G_{i,2k} \end{cases}$$
(5.3)

where  $b_{i,2k}$  denotes the value stored in the bitcell pair  $(G_{i,2k-1}, G_{i,2k})$ . Note that with this data storage scheme, the dimension of the computed dot product in a single row is  $\frac{N}{2}$ . Also, this scheme can be extended to bitcells storing multi-bit information in a straightforward manner.

In the absence of any variations and noise, one can derive an expression for the nominal SL current in *i*th row by applying Kirchhoff's current law at the SL to obtain (see Appendix A for a detailed derivation):

$$I_{\text{sig},i} = \left[\frac{R_{\text{arr},i}}{R_{\text{arr},i} + R_s}\right] \left(\sum_{k=1}^{N/2} V_{2k} \Delta G_{i,2k}\right) = S_I I_{\text{ideal},i}$$
(5.4)

where  $0 < S_I < 1$  (first term) is a scale factor,  $I_{\text{ideal},i}$  (second term) denotes ideal SL current when  $R_s = 0$ ,  $\Delta G_{i,2k} = G_{i,2k-1} - G_{i,2k}$ , and  $R_{\text{arr},i}$  (array resistance) is the Thevenin resistance looking into the *i*th SL given by:

$$R_{\text{arr},i} = \frac{1}{\sum_{j=1}^{N} G_{i,j}}$$
(5.5)

In the typical data storage scheme (see Eq. (5.3)), almost half bitcells store

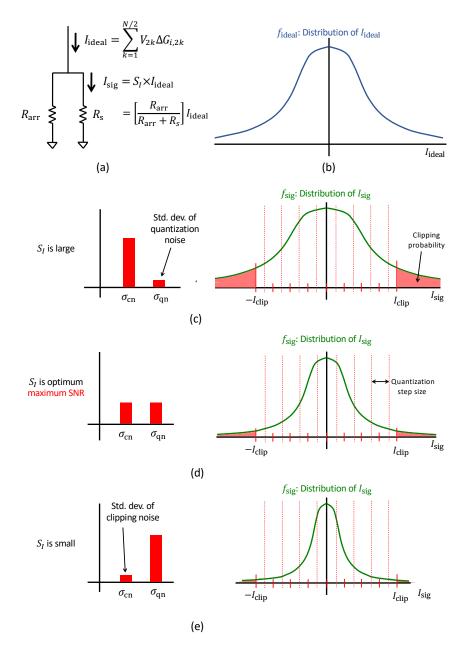


Figure 5.3: Interpreting Eq. (5.4): (a) equivalent circuit representation, (b) probability distribution of  $I_{\text{ideal}}$  and the probability distribution of  $I_{\text{sig}}$  when: (c)  $S_I$  is large ( $\sigma_{\text{cn}} \gg \sigma_{\text{qn}}$ ), (d)  $S_I$  is optimum ( $\sigma_{\text{cn}} \approx \sigma_{\text{qn}}$ ), and (e)  $S_I$  is small ( $\sigma_{\text{cn}} \ll \sigma_{\text{qn}}$ ), where  $\sigma_{\text{cn}}$  and  $\sigma_{\text{qn}}$  denote the standard deviations of clipping and quantization noise. The vertical dotted lines in (c)-(e) denote ADC quantization bins reflected to the input of the SL sensing circuit.

 $G_{\text{on}}$  and remaining store  $G_{\text{off}}$  in every row. Hence, the array resistance can be approximated as:

$$R_{\operatorname{arr},i} \approx \frac{1}{N\left(\frac{G_{\operatorname{on}}+G_{\operatorname{off}}}{2}\right)} = \frac{1}{NG_{\operatorname{avg}}} = \frac{R_{\operatorname{avg}}}{N}$$
(5.6)

where  $G_{\text{avg}} = \frac{G_{\text{on}}+G_{\text{off}}}{2} = \frac{R_{\text{avg}}}{N}$  denotes the average of two bitcell conductance states, and its inverse  $R_{\text{avg}}$  is referred to as *average bitcell resistance*.

Substituting the value of  $R_{\text{arr},i}$  from Eq. (5.6) in Eq. (5.4) and writing  $\Delta G_{i,2k} = b_{i,2k} \frac{1}{R_{\text{off}}} \left(\frac{R_{\text{off}}}{R_{\text{on}}} - 1\right)$  and  $V_{2k} = x_{2k} V_{\text{lsb}}$ , we get,

$$I_{\text{sig},i} \approx \left[\frac{R_{\text{avg}}}{R_{\text{avg}} + NR_s}\right] \left(\frac{R_{\text{off}}}{R_{\text{on}}} - 1\right) \left(\sum_{k=1}^{N/2} b_{i,2k} x_{2k}\right) \left(\frac{V_{\text{lsb}}}{R_{\text{off}}}\right)$$
(5.7)

where the term  $\frac{V_{\rm lsb}}{R_{\rm off}}$  is the minimum current through a bitcell. Notice that the third and fourth terms can be viewed as a numerical dot product scaled by the minimum bitcell current, while the first two terms are dimensionless gain factors. Naturally, in order to achieve high SNR, *both* of these factors need to be large. This shows that *ideally* bitcell resistance states should be such that their average  $R_{\rm avg} \gg NR_s$  and their contrast  $\frac{R_{\rm off}}{R_{\rm on}} \gg 1$  in order to achieve high SNR.

Equation (5.4) can be interpreted (see Fig. 5.3(a)) as describing the process of current division of  $I_{ideal}$  across two resistors in parallel, viz.  $R_s$  and  $R_{arr}$ , where  $I_{sig}$  flows through  $R_s$  and  $I_{ideal} - I_{sig}$  flows through  $R_{arr}$ . Since  $S_I$  linearly scales  $I_{sig}$ , its value controls the clipping vs. quantization noise trade-off. For example, a larger value of  $S_I$  results in higher clipping probability, and consequently a higher standard deviation of clipping noise  $\sigma_{nc}$ (see Fig. 5.3(b)). On the other hand, if  $S_I$  is too small, it squeezes the  $I_{sig}$ distribution close to zero, resulting in a higher standard deviation of quantization noise  $\sigma_{nq}$  (see Fig. 5.3(d)). Thus, there exists an optimum value of  $S_I$ when the clipping and quantization noise are balanced, *i.e.* when  $\sigma_{nc} \approx \sigma_{nq}$ (see Fig. 5.3(c)), which maximizes the SNR.

Notice that the value of  $S_I$  in Eq. (5.4) is dictated by both  $R_s$  and  $R_{arr}$  as follows:

• A smaller value of  $\frac{R_s}{R_{\text{arr},i}}$  results in a higher  $S_I$ . For example, when  $R_s \to 0$ , then  $S_I \to 1$  and  $I_{\text{sig},i} \approx I_{\text{ideal},i}$ . However, achieving such small

Parameter	Value
VDAC bit precision $(B_x)$	5 bit
ADC bit precision $(B_{adc})$	6 bit
$V_{ m lsb}$	$3\mathrm{mV}$
$I_{ m clip}$	$2\mu\mathrm{A}$
$\left(\frac{\sigma}{\mu}\right)_{\rm bc}$	4%
$\left(\frac{\sigma}{\mu}\right)_{V_{\text{lsb}}}$	4%

Table 5.1: Typical simulation parameter values

value of  $R_s$  (< 500 $\Omega$ ) can be very challenging due to significant area and energy overhead of the SL sensing circuits with low input resistance.

- Increasing the number of columns N decreases  $S_I$  since  $R_{\text{arr},i} \propto \frac{1}{N}$ . This limits the dot product dimension  $\frac{N}{2}$  that can be implemented.
- Smaller resistive contrast in the bitcell requires a smaller value of  $R_s$  for maintaining  $S_I$ . Thus, we can expect MRAM crossbar requiring a smaller  $R_s$  to achieve maximum SNR compared to RRAM and PCM.

Now we account for the impact of mismatch variations in the input voltage DACs by substituting  $V_{2k} \leftarrow V_{2k} + \delta V_{2k}$  in Eq. (5.4) and separating out the mismatch term to obtain

$$I_{\mathrm{dm},i} = \left[\frac{R_{\mathrm{arr},i}}{R_{\mathrm{arr},i} + R_s}\right] \left(\sum_{k=1}^{N/2} \delta V_{2k} \Delta G_{i,2k}\right) = S_I \left(\sum_{k=1}^{N/2} \delta V_{2k} \Delta G_{i,2k}\right)$$
(5.8)

where  $\delta V_{2k}$  denotes variation in the 2k-th DAC pair. Recall that we assume VDAC is designed using a current DAC followed by a TIA. Hence, its variations are dominated by the variations in the current DAC, *i.e.*  $\delta V_{2k} \sim \mathcal{N}\left(0, 2V_i V_{\text{lsb}}\left(\frac{\sigma}{\mu}\right)_{V_{\text{lsb}}}^2\right)$ , where  $\left(\frac{\sigma}{\mu}\right)_{V_{\text{lsb}}}$  corresponds to the deviation-to-mean ratio of the mismatch variations in a single current DAC finger. Similarly, we account for the impact of variations in the bitcell conductances by substituting  $\Delta G_{i,2k} \leftarrow \Delta G_{i,2k} + \delta G_{i,2k}$  in Eq. (5.4) and separating out the mismatch term to obtain

$$I_{\mathrm{bm},i} = \left[\frac{R_{\mathrm{arr},i}}{R_{\mathrm{arr},i} + R_s}\right] \left(\sum_{k=1}^{N/2} V_{2k} \delta G_{i,2k}\right) = S_I \left(\sum_{k=1}^{N/2} V_{2k} \delta G_{i,2k}\right)$$
(5.9)

where  $\delta G_{i,2k} \sim \mathcal{N}\left(0, \left(\frac{\sigma}{\mu}\right)^2_{\rm bc}(G_{\rm off}^2 + G_{\rm on}^2)\right)$  denotes the variation in the dif-

ference in conductances of 2k-th pair of bitcells,  $\left(\frac{\sigma}{\mu}\right)_{bc}$  denotes standard deviation-to-mean ratio for bitcell conductance mismatch. Also, notice that these conductance variations cause variations in  $R_{arr,i}$ . However, it only changes the range of  $I_{sig,i}$  independent of the input vector  $\mathbf{x}$ . Hence, we do not include it in our analysis in this chapter since it can be accounted for by a corresponding small change in the ADC dynamic range. Note in Eq. (5.8) and Eq. (5.9) that both  $I_{bm,i}$  and  $I_{dm,i}$  scale with  $S_I$  in the same manner as  $I_{sig}$ . Hence, the ratios  $\frac{I_{sig}}{I_{bm,i}}$  and  $\frac{I_{sig}}{I_{dm,i}}$  do not depend on  $S_I$ , and hence on  $R_s$ . The implications will become clear in Sec. 5.3.1 where we will see that two of the SNR regimes are primarily dictated by the other two noise, *i.e.* quantization and clipping noise.

Next, we employ the expressions in Eq. (5.8) and Eq. (5.9) to study the impact of  $R_s$ , N as well as the BCA energy per operation for all three memory types on the SNR. Note that the expectations in SNR are estimated empirically via Monte Carlo simulations. The typical values of parameters chosen for these simulations are given in Table 5.1. Also, we constrain  $b_{i,2k} \in \{-1, 1\}$ in our simulations for simplicity.

#### 5.3.1 SNR vs. $R_s$ Trade-off

We begin with a study of SNR vs.  $R_s$  trade-off. Given the tight area constraints on the SL sensing circuits, achieving very low  $R_s$  could be prohibitively expensive in resistive IMC crossbars. Hence it is important to understand how SNR changes with  $R_s$  and be able to estimate the value of  $R_s$  which maximizes the SNR. Figures 5.4(a), (b), and (c) show SNR vs.  $R_s$ trade-off for VDAC driven crossbars consisting of PCM, RRAM, and MRAM memory devices, respectively. We make following observations from Fig. 5.4:

- For each value of N, SNR is maximized at a particular value of  $R_s$  (defined as  $R_s^*$ ) for all three memory device technologies. Furthermore, we observe that  $R_s^*$  decreases with N. This is expected since  $R_{\rm arr}$  decreases with N.
- Recall that  $I_{sig}$  decreases when  $R_s$  increases. Thus, when  $R_s < R_s^*$ , clipping noise dominates, while quantization noise dominates when  $R_s > R_s^*$ .

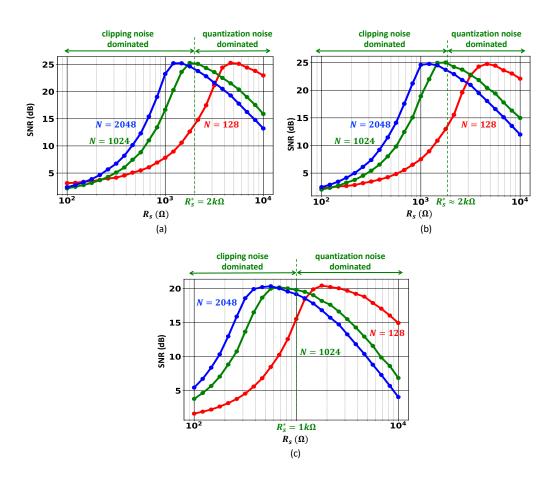


Figure 5.4: SNR vs.  $R_s$  trade-off in IMC with VDAC driven crossbars consisting of (a) PCM, (b) RRAM, and (c) MRAM memory devices.

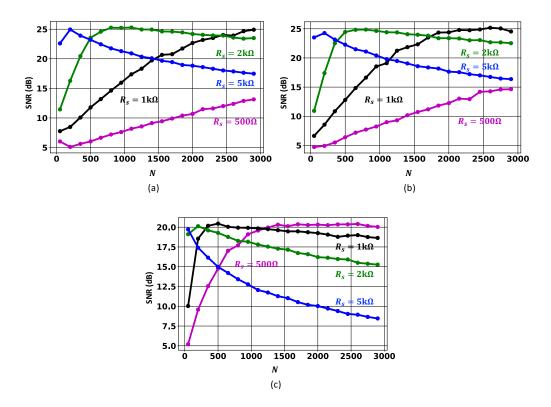


Figure 5.5: SNR vs. N trade-off in IMC with VDAC driven crossbars consisting of (a) PCM, (b) RRAM, and (c) MRAM memory devices.

• For a given value of N,  $R_s^*$  also depends weakly upon the resistive contrast of the memory device. For example, with N = 1024,  $R_s^*$  for MRAM is smaller than that of PCM and RRAM due to smaller resistive contrast of MRAM.

Notice that  $R_s^* \approx 1 \,\mathrm{k}\Omega$ -to- $2 \,\mathrm{k}\Omega$  for all three memory technologies, which is feasible even with the pitch-matching contraints of IMCs.

#### 5.3.2 SNR vs. N Trade-off

Next, we now study the SNR vs. N trade-off for distinct values of  $R_s$ . Figures 5.5(a), (b), and (c) show the SNR vs. N trade-off for distinct values of  $R_s$  for VDAC driven crossbars consisting of PCM, RRAM, and MRAM memory devices, respectively. For RRAM and PCM, when  $R_s$  is too small (*e.g.*  $R_s \leq 1 \,\mathrm{k}\Omega$ ), SNR increases with N since it remains dominated by clipping noise until N = 3000. For larger  $R_s$  values (*e.g.*  $2 \,\mathrm{k}\Omega \leq R_s \leq 5 \,\mathrm{k}\Omega$ ), SNR first increases with N until  $N = N^*$  and then decreases when  $N > N^*$ , where it remains dominated by quantization noise. For example, with  $R_s = R_s^* = 2 \,\mathrm{k}\Omega$ ,  $N^* \approx 1000$  and  $N^* \approx 750$  for PCM and RRAM, respectively. Also, notice that the decrease in SNR as a function of N in this case is very gradual for both PCM and RRAM. This is due to the opposing influence of two effects (see Eq. (5.4)): (i)  $\mathcal{O}(\frac{1}{N})$  decrease in  $I_{\mathrm{sig}}$  via  $R_{\mathrm{arr}}$ , and (ii)  $\mathcal{O}(\sqrt{N})$  increase in  $I_{\mathrm{sig}}$  on average via  $\sum_{k=1}^{N/2} V_{2k} \Delta G_{i,2k}$ . Thus, SNR remains high for a broad range of N values.

Similar observations can be made for MRAM in Fig. 5.5(c). For instance, high SNR is achieved at a broad range of  $500 \leq N \leq 1000$  with  $R_s = R_s^* = 1 \text{ k}\Omega$ . Also note that the maximum value of SNR for MRAM is only 20 dB, which is 5 dB smaller than that for PCM/RRAM. This is a direct consequence of MRAM's smaller resistive contrast.

#### 5.3.3 Bound on M

Increasing M in IMC crossbars linearly improves throughput via parallelism. Naturally one would want to choose as high M as possible. Hence it is important to address following question: "How large can the number of rows (M) be in VDAC driven crossbars?". We address it in this subsection.

The value of M is limited by the constraint that  $I_{\rm BL,max} < I_{\rm VDAC,max}$ , where  $I_{\rm VDAC,max}$  denotes the maximum current that can be provided by VDAC while preserving its voltage. Based on Eq. (5.4), we first derive an expression for the maximum BL current  $I_{\rm BL,m}$  in a single column in a given crossbar with the sensing circuit resistance  $R_s$  by summing the maximum individual bitcell currents along the column to obtain:

$$I_{\rm BL,max} = \sum_{i=1}^{M} \left[ \frac{R_{\rm arr,i}}{R_{\rm arr,i} + R_s} \right] V_{\rm max} G_{\rm on}$$
(5.10)

where  $V_{\text{max}}$  maximum output voltage of VDAC. Note that a VDAC needs to provide the maximum current when all bitcell in its column are ON and its input corresponds to the  $V_{\text{max}}$ . For a  $B_x$  bit VDAC,  $V_{\text{max}} = 2^{B_x} V_{\text{lsb}}$ .

As observed in Eq. (5.10),  $I_{\rm BL,max}$  increases linearly with M. We plot  $I_{\rm BL,max}$  vs. M for N = 200 and N = 1000 in Fig. 5.6 for all three memory device technologies. For RRAM/PCM, we choose  $R_s^* = 5 \,\mathrm{k}\Omega \,(R_s^* = 2 \,\mathrm{k}\Omega)$  corresponding to  $N = 200 \,(N = 1000)$  (recall from Fig. 5.5). Similarly, for

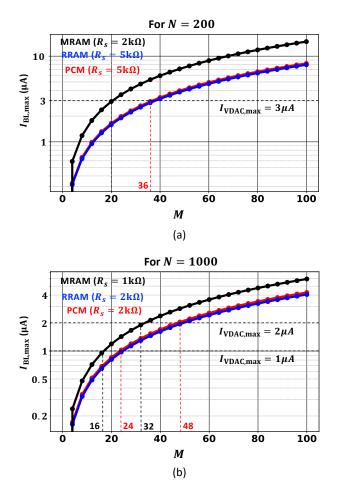


Figure 5.6:  $I_{\rm BL,max}$  vs. M trade-off in IMC with voltage driven crossbars.

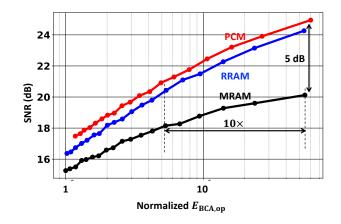


Figure 5.7: SNR vs. normalized BCA energy trade-off in IMC with VDAC driven crossbars consisting of PCM, RRAM, and MRAM memory devices.

MRAM, we choose  $R_s^* = 2 k\Omega$  and  $R_s^* = 1 k\Omega$  corresponding to N = 200 and N = 1000, respectively.

Note that given the stringent area constraints in the IMC crossbars,  $I_{\text{VDAC,max}}$ of a couple of  $\mu$ A is feasible to achieve. With  $I_{\text{VDAC,max}} = 2 \,\mu$ A ( $I_{\text{VDAC,max}} = 1 \,\mu$ A), we find that  $M = 48 \,(M = 24)$  satisfies the constraint for RRAM/PCM with N = 1000 (see Fig. 5.6(b)). For MRAM, corresponding M values are reduced to 32 and 16, respectively. For smaller value of N = 200, we find that  $I_{\text{BL,max}}$  is higher for the similar range of M values for all three memory device technologies. Specifically, with larger  $I_{\text{VDAC,max}} = 3 \,\mu$ A, we get M = 36 and M = 20 for RRAM/PCM and MRAM, respectively (see Fig. 5.6(a)).

#### 5.3.4 SNR vs. Energy Trade-off

IMCs have a fundamental trade-off between the SNR and energy consumption, which determines their applicability in different tasks [65]. Figure 5.7 plots SNR vs.  $E_{\rm BCA,op}$  trade-off for PCM, RRAM, and MRAM, where  $E_{\rm BCA,op}$  denotes the BCA energy per 1 bit×1 bit MAC operation consumed during steady-state IMC operation. We estimate  $E_{\rm BCA,op}$  as

$$E_{\rm BCA,op} = \frac{1}{N} \frac{1}{B_x} V_{\rm DD} T_{\rm BCA} \mathbb{E} \left[ |I_{\rm sig}| \right]$$
(5.11)

where  $V_{\text{DD}}$ ,  $T_{\text{BCA}}$  denote supply voltage and the ON duration of BL DACs, respectively, and  $|\cdot|$  denotes the absolute value operation. We assume  $V_{\text{DD}}$ ,  $T_{\text{BCA}}$  to be same across the three memory types: RRAM, PCM, and MRAM.

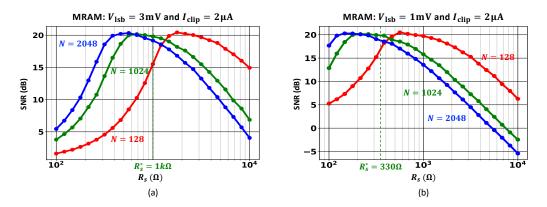


Figure 5.8: SNR vs.  $R_s$  trade-off in IMC MRAM crossbars with  $I_{clip} = 2 \mu A$  and: (a)  $V_{lsb} = 3 \text{ mV}$ , and (b)  $V_{lsb} = 1 \text{ mV}$ .

In Fig. 5.7, N is swept to obtain different points in the plot and  $E_{\rm BCA,op}$  values are normalized with respect to the smallest  $E_{\rm BCA,op}$  for MRAM (corresponding to N = 1950). For comparable  $E_{\rm BCA,op}$ , SNR for MRAM crossbar is 5 dB smaller than that for RRAM/PCM crossbars. As we also discussed in Sec. 5.3.2, this is a direct consequence of smaller resistive contrast in MRAM. Figure 5.7 also quantifies the energy penalty of small resistive contrast. Specifically, RRAM and PCM achieve  $\approx 10 \times$  smaller  $E_{\rm BCA,op}$  compared to MRAM for comparable SNR of 20 dB. The larger separation between  $R_{\rm on}$  and  $R_{\rm off}$  values in RRAM/PCM enables their operation at smaller bitcell currents compared to MRAM while achieving similar computation accuracy.

#### 5.3.5 Impact of $V_{\rm lsb}$ and $I_{\rm clip}$

In earlier sections (from Sec. 5.3.1 to Sec. 5.3.4), we assumed  $V_{\rm lsb} = 3 \,\mathrm{mV}$ and  $I_{\rm clip} = 2 \,\mu \mathrm{A}$ . While they can be reasonably realized in advanced process technologies, in this section, we discuss how the SNR trade-off will change w. r. t. both  $V_{\rm lsb}$  and  $I_{\rm clip}$ . We focus particularly on MRAM in this section since it turns out to be the most challenging due to its lower resistive contrast. However, similar conclusions can also be drawn for RRAM and PCM.

Figure 5.8 shows the impact of reducing  $V_{\rm lsb}$  on SNR vs.  $R_s$  trade-off. Specifically, we find that a reduction in  $V_{\rm lsb}$  results in a proportionate reduction in  $R_s^*$ . For example, in Fig. 5.8, changing  $V_{\rm lsb}$  from 3 mV to 1 mV results in a proportionate 3× reduction in  $R_s^*$ . This is expected since  $V_{\rm lsb}$ 

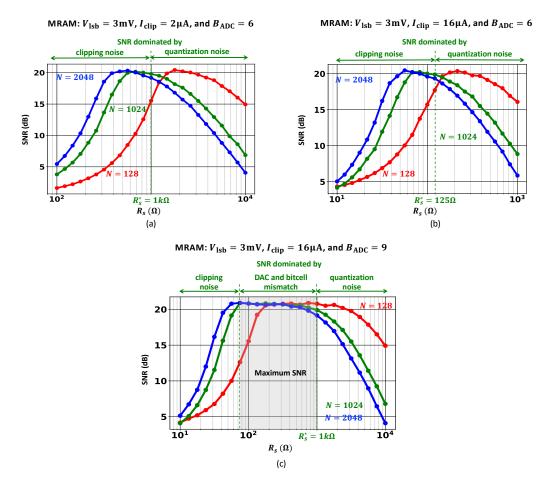


Figure 5.9: SNR vs.  $R_s$  trade-off in IMC MRAM crossbars with: (a)  $I_{\text{clip}} = 2 \,\mu\text{A}$  and  $B_{\text{adc}} = 6 \,\text{bit}$ , (b)  $I_{\text{clip}} = 16 \,\mu\text{A}$  and  $B_{\text{adc}} = 6 \,\text{bit}$ , and (c)  $I_{\text{clip}} = 16 \,\mu\text{A}$  and  $B_{\text{adc}} = 9 \,\text{bit}$ .

linearly changes  $I_{\text{ideal}}$  in Eq. (5.4). Hence, the  $S_I$  needs to be correspondingly adjusted via a proportionate change in the  $R_s$  for maintaining the same SNR.

Similarly, we explore the impact of  $I_{\text{clip}}$  on SNR vs.  $R_s$  trade-off in Fig. 5.9. Note that the choice of  $I_{\text{clip}}$  is closely linked with the ADC bit precision  $B_{\text{adc}}$ since the quantization noise  $I_{\text{qn}} \sim U\left(-\frac{I_{\text{clip}}}{2^{B_{\text{adc}}}}, \frac{I_{\text{clip}}}{2^{B_{\text{adc}}}}\right)$  (recall from Sec. 5.2). Hence, we consider following two cases in Fig. 5.9:  $I_{\text{clip}}$  is increased  $8 \times$  from  $2 \,\mu\text{A}$  to  $16 \,\mu\text{A}$  (1) while preserving  $B_{\text{adc}} = 6$  bit (Fig. 5.9(b)), and (2) while proportionately increasing  $B_{\text{adc}}$  from 6 bit to 9 bit in order to accommodate the larger current range (Fig. 5.9(c)). In Fig. 5.9(a) and Fig. 5.9(b), we find that when  $I_{\text{clip}}$  is increased while maintaining the  $B_{\text{adc}}$ , the same SNR vs.  $R_s$  trade-off is achieved at lower  $R_s$  values. Higher  $I_{\text{clip}}$  increases quantization noise, which needs to be compensated by appropriately increasing  $S_I$ 

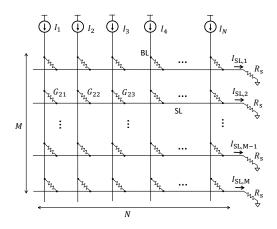


Figure 5.10: Current driven crossbar.

via proportionate reduction in  $R_s$ . Thus, the optimal value  $R_s^*$  is proportionately lower at the higher  $I_{\text{clip}}$ . The scenario gets interesting when  $B_{\text{adc}}$  is increased along with  $I_{\text{clip}}$ . In Fig. 5.9(c), there is a broad range of  $R_s$  (from  $80 \Omega$  to  $1 \text{ k}\Omega$ ) where the SNR is maximized. In this regime, both quantization and clipping noise are small and the SNR is dominated by DAC and bitcell conductance mismatch variations. Note that the impact of these mismatch variations in Eq. (5.9) and Eq. (5.8) also gets scaled by the  $S_I$  making the SNR independent of  $R_s$  when these variations are dominant. Thus, higher  $I_{\text{clip}}$  and  $B_{\text{adc}}$  enable a broader design space by reducing clipping noise without any increase in the quantization noise.

## 5.4 Current Driven Crossbars

Figure 5.10 shows current driven crossbar, where each BL current  $I_i = x_i I_{\rm lsb}$ and  $I_{\rm lsb}$  corresponds to the change in DAC current corresponding to the LSB. Unlike voltage driven crossbars, deriving a closed-form analytical expression for  $I_{{\rm SL},i}$  is mathematically intractable for an arbitrary value of M. Hence, we analyze a  $2 \times N$  crossbar array. Doing so provides very good insights regarding the challenges involved in current DAC driven crossbars.

In the absence of any variations or noise, we apply the Kirchhoff's current law at the *i*th SL along with the charge conservation principle to obtain the SL current for  $2 \times N$  crossbar array ( $i \in \{1, 2\}$ ) as follows (see Appendix B for a detailed derivation):

$$I_{\text{sig},i} = \left[\frac{R_s}{R_{\text{arr}} + 2R_s}\right] \left(\sum_{j=1}^N I_j\right) + \left[\frac{R_{\text{arr}}}{R_{\text{arr}} + 2R_s}\right] \left(\sum_{j=1}^N \frac{G_{1j}I_j}{G_{1j} + G_{2j}}\right) \quad (5.12)$$

$$=\beta_{\rm cm}I_{\rm tot} + \beta_{\rm dp}I_{\rm dp} \tag{5.13}$$

$$=\frac{1}{2}(1-\beta_{\rm dp})I_{\rm tot}+\beta_{\rm dp}I_{\rm dp}$$
(5.14)

where  $\beta_{dp}$  corresponds to the scaling of  $I_{sig,i}$  component that is proportional to the dot product (signal component),  $\beta_{cm}$  corresponds the scaling of the constant bias current component (bias component), and  $R_{arr}$  denotes the equivalent resistance corresponding to columnwise parallel combination of all bitcells as follows:

$$\frac{1}{R_{\rm arr}} = \sum_{j=1}^{N} \frac{1}{R_{1,j} + R_{2,j}}$$
(5.15)

Notice that the SL currents in current DAC driven crossbars differ from those in VDAC driven crossbars in the following two ways:

- 1. Need for constant sum of bitcell conductances along BL: Each elementwise multiplication in  $I_{dp}$  is scaled by the sum of bitcell conductances along that particular BL. Hence, this sum needs to be constant across columns in order to achieve an accurate dot product. As discussed in Sec. 4.5.4, this sum can be equalized along columns by incorporating redundant BCA rows.
- 2. Large input-dependent bias current: Notice that the first term  $(\frac{1}{2}(1-\beta_{dp})I_{tot})$  in Eq. (5.12) corresponds to an input vector **x** dependent bias current that is identical across all rows. Notice that this term appears due to the non-zero input impedance of sensing circuits, *i.e.*  $R_s \neq 0$ . Since  $R_{arr}$  and hence  $\beta_{dp}$  decreases with N, the relative proportion of this bias current increases with N. For example, with N = 200 and  $R_s^* = 5 \,\mathrm{k\Omega} \,(2 \,\mathrm{k\Omega})$  for PCM/RRAM (MRAM), we have:
  - PCM:  $\beta_{dp} = 0.603, \ \frac{\beta_{dp}}{\beta_{cm}} \approx 304\%$
  - RRAM:  $\beta_{dp} = 0.140, \frac{\beta_{dp}}{\beta_{cm}} \approx 32.5\%$
  - MRAM:  $\beta_{dp} = 0.004$ ,  $\frac{\beta_{dp}}{\beta_{cm}} \approx 0.9\%$

In PCM, thanks to its large  $R_{\text{off}}$  values,  $\beta_{dp}$  remains  $3 \times$  higher than  $\beta_{\text{cm}}$ . Even in RRAM,  $\beta_{dp}$  and  $\beta_{\text{cm}}$  remain comparable. However, in MRAM,  $\beta_{dp}$  turns out to be only 0.9% of  $\beta_{\text{cm}}$  due to its smaller  $R_{\text{on}}/R_{\text{off}}$  values. Thus, a significant portion of  $I_{\text{sig},i}$  corresponds to the constant bias current. Also, a larger value of  $\beta_{\text{cm}}$  corresponds to an amplified impact of the mismatch variations in the current DACs on  $I_{\text{sig},i}$ . Specifically, the impact of DAC mismatch variations,  $I_{\text{dm},i}$  (recall Sec. 5.2), in current driven crossbars is given as:

$$I_{\mathrm{dm},i} = \beta_{\mathrm{cm}} \left(\sum_{j=1}^{N} \delta I_j\right) + \beta_{\mathrm{dp}} \left(\sum_{j=1}^{N} \frac{G_{1j} \delta I_j}{G_{1j} + G_{2j}}\right)$$
(5.16)

where  $\delta I_j$  denotes the variation in *j*th current DAC. Notice that  $\beta_{\rm cm}$  also scales the sum of total variations in the current DACs (first term in Eq. (5.16)). Thus, when  $\beta_{\rm cm} \gg \beta_{\rm dp}$  (as is in the case of MRAM), the impact of mismatch variations in DACs will be larger than the signal current, *i.e.*  $I_{{\rm dm},i} \gg \beta_{\rm dp} I_{\rm dp}$ .

Challenge 2 discussed above turned out to be the key difficulty in our chip prototype of current driven MRAM-DIMA in Sec. 4.6. In our prototype, it did turn out that  $\beta_{\rm cm} \gg \beta_{\rm dp}$ , resulting in SL currents getting overwhelmed by the signal-independent bias current  $\beta_{\rm cm} \left(\sum_{j=1}^{N} \delta I_j\right)$ , resulting in a very low SNR during the chip measurements of current driven MRAM-DIMA prototype. This being the fundamental device-level limitation in the current drive setting, we concluded that one needs to employ on-chip learning based compensation to accurately recover  $\beta_{\rm dp}I_{\rm dp}$  in the presence of DAC mismatch variations in the current driven MRAM crossbars.

### 5.5 Discussion

In this chapter, we analyzed the SNR of resistive crossbar arrays employed for in-memory computing. The analysis is device-agnostic and we study three memory device types, namely PCM, RRAM, and MRAM, having distinct resistive contrast. We find that there are optimal values sensing impedance and the number of columns that achieve maximum SNR in the case of voltage driven crossbars. We also considered current driven crossbars and identified their challenges compared to VDAC driven crossbars. Based on our analysis, we draw the following four key conclusions:

- Voltage drive works well for IMC with resistive crossbars for all three memory device technologies (PCM, RRAM, and MRAM), despite large differences in their resistive contrast. The highest SNR is achieved with  $R_s \approx 2 \,\mathrm{k}\Omega$  for N = 1000, which is certainly feasible even after stringent area constraints. Furthermore, with  $M \approx 20$  and  $N \approx 1000$ , one can reap the benefits of IMC while maintaining maximum DAC current reasonably low (< 1  $\mu$ A).
- Current drive has a key disadvantage of large impact of DAC mismatches compared to voltage drive. It is particularly exacerbated in the case of MRAM due to its low resistive contrast and relatively large sensing impedance  $R_s$ . While reducing  $R_s$  by orders of magnitude will alleviate this challenge, it is often prohibitively expensive in terms of area and energy. Since the first term in Eq. (5.16) depends only on the inputs and is same for all SLs, one can compensate it via a shared circuit.
- MRAM achieves significantly poorer SNR vs. energy trade-off compared to RRAM and PCM (see Fig. 5.7), primarily due to its lower resistive contrast. Since this is a fundamental device limitation, it underscores the need to employ on-chip learning [65] based statistical error compensation (SEC) for improving SNR in MRAM based IMC.
- Equation (5.7) shows that ideally the bitcell resistance states should have their average  $R_{\text{avg}} \gg NR_s$  and their contrast  $\frac{R_{\text{off}}}{R_{\text{on}}} \gg 1$  in order to achieve high SNR. Furthermore, given  $E_{\text{BCA,op}} \propto I_{\text{sig}}$ , one can trade-off the SNR improvements obtained via high average bitcell resistance and high resistive contrast to appropriately minimize the energy consumption. For example, higher  $R_{\text{avg}}$  and  $\frac{R_{\text{off}}}{R_{\text{on}}}$  can enable smaller sensing circuit impedance  $R_s$  and higher ADC quantization step size, simplifying the sensing circuit design for the same target SNR.

Moving forward, an important next step is to characterize the impact of BL/SL parasitic impedance in crossbars on the SNR of in-memory computation. Also, in this chapter, we focus on crossbar arrays, where BLs are perpendicular to SL. However, our approach can be followed to carry out similar analysis for parallel-bar arrays where BLs are parallel to SL.

## CHAPTER 6

# SUBSPACE ANALYSIS OF ADVERSARIAL PERTURBATIONS

### 6.1 Overview

Deep nets are vulnerable to *adversarial perturbations* [9], *i.e.*, changes imperceptible to humans result in incorrect decisions. Multiple approaches to construct adversarial examples (attack methods) have been proposed [28, 11, 29, 12, 30]. While the basic idea is to move the input in the direction of increasing loss, stronger attacks [13, 12] operate iteratively by perturbing the input in small increments. On the defense side, early attempts to provide adversarial defenses obfuscate gradient computation by introducing non-differentiable and/or randomized operations in the forward pass [35, 36, 37, 38]. However, such defenses were shown to be ineffective [42, 14], especially when the attacker employs adaptive attacks, *i.e.*, the attacks that modified specifically to evade a given defense. Today adversarial training (AT) [13, 22, 43] is the only empirical defense that has remained robust to a wide variety of the strongest possible known attacks.

Thus, defending deep nets against adversarial perturbations remains a formidable challenge. This is in part due to a lack of in-depth understanding of the underlying cause of deep nets' vulnerability to adversarial perturbations. Indeed, efforts to improve such understanding have led to intriguing observations/insights such as the role of the geometry of the decision boundary [187, 188], and the existence of an inherent trade-off between generalization and robustness [22, 55], the existence of universal perturbations [189], and the notion of robust vs. non-robust features [190]. Such insights, however, do not yet completely explain the cause of a DNN's adversarial vulnerability. Hence, further efforts are required to deepen our understanding of this phenomenon so that effective defenses against adversarial attacks can be developed. In this chapter, we employ subspace analysis to understand the geometric orientations of adversarial perturbations, and how they differ after vanilla vs. adversarial training. Specifically, we present and validate two hypotheses about geometric orientations of adversarial perturbations. In Sec. 6.2, we observe how adversarial perturbations are oriented relative to dominant image subspace and vice versa. In Sec. 6.3, we explore how orientations perturbations of different types, namely  $\ell_{\infty}$ ,  $\ell_2$ , and  $\ell_1$  perturbations, differ from each other in terms of their dominant subspaces. Note that the eventual goal of analysis presented in this chapter is to develop insights for improving robustness vs. cost trade-off in deep nets. We discuss the implications of our analysis in Sec. 6.4.

Without loss of generality, we employ ResNet-18 network on CIFAR-10 for the subspace analysis in this chapter. The analysis can be repeated for any other network or dataset. We refer to deep net parameterized by  $\theta$  as  $f_{\theta}(\boldsymbol{x})$ , and the training set as  $X = \{\boldsymbol{x}_i\}_{i=1}^R$  with cardinality |X| = R and  $\boldsymbol{x}_i \in \mathbb{R}^D$ . For the CIFAR-10 dataset R = 50,000 and D = 3072. We employ vanilla training to obtain a non-robust network  $f_{\theta}^{\text{van}}$ , while use TRADES [22] training (one of the SOTA defense) to obtain a robust network  $f_{\theta}^{\text{rob}}$ .

## 6.2 Orthogonality between Images and Adversarial Perturbations

We begin by a following hypothesis (briefly alluded to in [188]): adversarial perturbations of vanilla trained networks lie predominantly in a subspace orthogonal to that occupied by the input. Conversely, an adversarially trained network will find its adversarial perturbations to lie in the same subspace as the input. We refer to this hypothesis as Subspace Orthogonality of Adversarial Perturbations (SOAP).

We employ a subspace analysis to empirically validate the SOAP hypothesis of adversarial perturbations. For both robust and non-robust deep nets we generate adversarial perturbations via DeepFool [11], *i.e.*, we iteratively find a perturbation vector  $\boldsymbol{\delta}^{\alpha} \in \mathbb{R}^{D} \ \forall \alpha \in \{\text{van, rob}\}$  as an approximate solution

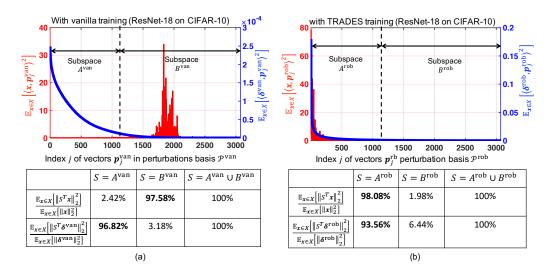


Figure 6.1: Dominant adversarial perturbation subspaces in robust vs. nonrobust networks: (top plots) average squared projections of training images  $\mathbb{E}_{\boldsymbol{x}\in X}[\langle \boldsymbol{x}, \boldsymbol{p}_j^{\alpha} \rangle^2]$  and DeepFool perturbations  $\mathbb{E}_{\boldsymbol{x}\in X}[\langle \boldsymbol{\delta}^{\alpha}, \boldsymbol{p}_j^{\alpha} \rangle^2]$  on the perturbation basis  $P^{\alpha}$  for (a) non-robust ResNet-18  $\alpha$  = van, and (b) robust ResNet-18  $\alpha$  = rob, where  $\langle \cdot \rangle$  denotes dot product. (bottom tables) Fraction of average squared norm within the two orthogonal subspaces  $A^{\alpha} = \{\boldsymbol{p}_1^{\alpha}, \ldots, \boldsymbol{p}_{1200}^{\alpha}\}, B^{\alpha} = \{\boldsymbol{p}_{1201}^{\alpha}, \ldots, \boldsymbol{p}_{3072}^{\alpha}\}$ , together spanning entire perturbation basis  $P^{\alpha}$  for (a) non-robust network  $\alpha$  = van, and (b) robust network  $\alpha$  = rob, where  $S^T \boldsymbol{a}$  denotes projection of vector  $\boldsymbol{a}$  on subspace S.

to the problem:

$$\arg\min_{\boldsymbol{\delta}^{\alpha}} \|\boldsymbol{\delta}^{\alpha}\|_{2} \quad \text{s.t.} \quad f^{\alpha}_{\theta}(\boldsymbol{x} + \boldsymbol{\delta}^{\alpha}) \neq f^{\alpha}_{\theta}(\boldsymbol{x}), \alpha \in \{\text{van, rob}\}$$
(6.1)

Note that DeepFool perturbations are the vectors with approaximately the smallest  $\ell_2$  magnitude. Hence, they can be viewed as directions to the nearest decision boundary for their corresponding inputs.

Further, following [188], we compute the singular vector basis  $\mathcal{P}^{\alpha} = \{\boldsymbol{p}_{1}^{\alpha}, \ldots, \boldsymbol{p}_{D}^{\alpha}\}$  for the set of adversarial (DeepFool) perturbations  $\Delta^{\alpha} = \{\boldsymbol{\delta}_{1}^{\alpha}, \ldots, \boldsymbol{\delta}_{D}^{\alpha}\}$  $\forall i, \alpha \in \{\text{van, rob}\}$ . Again, the singular vectors  $\boldsymbol{p}_{i}^{\alpha}$  are ordered in descending order of the corresponding singular values. We refer to the basis  $\mathcal{P}^{\alpha}$  as the *perturbation basis*.

Figure 6.1(a) shows the mean squared projections of the images  $\boldsymbol{x}_i$  (red) and DeepFool perturbations  $\boldsymbol{\delta}_i^{\text{van}}$  (blue) onto the perturbation basis  $\mathcal{P}^{\text{van}}$ , averaged over the dataset X. We find that the adversarial perturbations for  $f_{\theta}^{\text{van}}$  (blue) lie in a subspace that is (mostly) *orthogonal* to that of the dominant image subspace (red). Indeed, it can be seen that these adversarial perturbations exploit dimensions with small input projections. One might conjecture that the dimensions having small input projections may not be critical for a classification task. However, [188] demonstrates that the small image projections on the dominant singular vectors  $\mathcal{P}^{\text{van}}$  do in fact contribute significantly to the network's natural accuracy. Thus, small perturbations in those dimensions cause a catastrophic drop in the network's robust accuracy. In contrast, for the robust deep net  $f^{\text{rob}}_{\theta}$ , Fig. 6.1(b) indicates that the adversarial perturbations and images predominantly lie in the *same* subspace.

We quantify this orthogonality (SOAP) property by computing the ratio of the squared norm of the projections in the two mutually orthogonal subspaces  $A^{\alpha} = \{\boldsymbol{p}_{j}^{\alpha}\}_{j=1}^{1200}$  and  $B^{\alpha} = \{\boldsymbol{p}_{j}^{\alpha}\}_{j=1201}^{3072}$  of basis  $\mathcal{P}^{\alpha} = A^{\alpha} \bigcup B^{\alpha}$ . Hence, for a given image  $\boldsymbol{x}_{i}$ ,

$$\|\boldsymbol{x}_i\|_2^2 = \|(A^{\alpha})^T \boldsymbol{x}_i\|_2^2 + \|(B^{\alpha})^T \boldsymbol{x}_i\|_2^2$$
(6.2)

where  $S^T \boldsymbol{a}$  denotes the projection of vector  $\boldsymbol{a}$  onto the subspace S, and  $\|\cdot\|_2$ denotes  $\ell_2$  norm. Hence, the ratio  $\|(A^{\alpha})^T \boldsymbol{x}_i\|_2^2 / \|\boldsymbol{x}_i\|_2^2$  indicates the extent to which a given vector  $\boldsymbol{x}_i$  lies in the subspace  $A^{\alpha}$ . Note that  $A^{\alpha}$  is the dominant subspace for the DeepFool perturbations since the singular vectors are ordered in the descending order of the corresponding singular values.

The table in Fig. 6.1(a) shows that the vanilla network  $f_{\theta}^{\text{van}}$  has 96.82% of the average squared  $\ell_2$  norm of the perturbations  $\delta_i$  lying in  $A^{\text{van}}$  while 97.58% of the average squared  $\ell_2$  norm of the training images lie in the orthogonal subspace  $B^{\text{van}}$ . Thus, the DeepFool perturbations and the training images lie almost entirely in mutually orthogonal subspaces. In contrast, Fig. 6.1(b) shows that the adversarially trained robust network  $f_{\theta}^{\text{rob}}$  has 93.56% of the average squared  $\ell_2$  norm of the adversarial perturbations and 98.08% of that of the training images lie in the same subspace  $A^{\text{rob}}$ . Thus, both images and the DeepFool perturbations for a robust network predominantly lie in the same subspace.

The distinction between non-robust and robust networks observed in Fig. 6.1 corroborates the generative behavior of adversarial perturbations of robust networks [190, 43]: adversarial perturbations of robust networks exhibit semantics similar to images because they predominantly lie in the same subspace. In the case of non-robust networks, the perturbations remain seman-

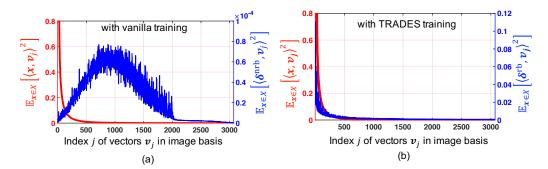


Figure 6.2: Average squared projections of training images  $\mathbb{E}_{\boldsymbol{x}\in X}[\langle \boldsymbol{x}, \boldsymbol{v}_j \rangle^2]$ and DeepFool perturbations  $\mathbb{E}_{\boldsymbol{x}\in X}[\langle \boldsymbol{\delta}^{\alpha}, \boldsymbol{v}_j^{\alpha} \rangle^2]$  on the singular vector basis  $\{\boldsymbol{v}_1, \ldots, \boldsymbol{v}_{3072}\}$  of training images for: (a) ResNet-18 trained with standard training ( $\alpha = \text{van}$ ), and (b) ResNet-18 trained with TRADES training, ( $\alpha = \text{rob}$ ).

tically uninterpretable because of their orthogonality to the image subspace.

The distinction between the alignment of adversarial perturbations and the images for  $f_{\theta}^{\text{van}}$  and  $f_{\theta}^{\text{rob}}$  is also observed when projecting onto the image basis  $\mathcal{V} = \{v_1, \ldots, v_D\}$  as shown in Fig. 6.2. The image basis  $\mathcal{V}$  is obtained via a singular vector decomposition of the training images and vectors  $v_i$ in  $\mathcal{V}$  are ordered in descending order of the corresponding singular values. Figure 6.2(a) shows the orthogonality of the DeepFool perturbations observed after vanilla training, while Fig. 6.2(b) shows the collinearity between the perturbation and image subspaces after TRADES training.

## 6.3 Distinction between Different Perturbation Models

In this section, we employ subspace methods to comprehend the distinction between  $\ell_{\infty}$ ,  $\ell_2$  and  $\ell_1$  perturbation types. For each input  $\boldsymbol{x}_i \in \mathbb{R}^D$  in dataset X, consider adversarial perturbations  $\boldsymbol{\alpha}_i$ ,  $\boldsymbol{\beta}_i$ , and  $\boldsymbol{\gamma}_i$  bounded within  $\ell_{\infty}$ ,  $\ell_2$ , and  $\ell_1$  norms, respectively. We can employ standard PGD attacks to obtain these perturbations.

We begin with the following hypothesis illustrated in Fig. 6.3: The perturbations  $\alpha$ ,  $\beta$ , and  $\gamma$  corresponding to input x have directions that differ significantly if the curvature of the decision boundary is high in the neighborhood of x. Conversely, if the curvature of the decision boundary is low, the perturbations  $\alpha$ ,  $\beta$ , and  $\gamma$  tend to lie in similar directions.

In fact, the curvature of the decision boundary of deep nets has been stud-

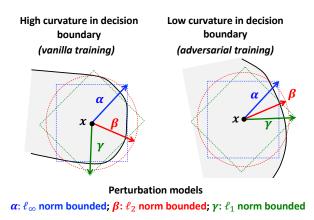


Figure 6.3: Illustration of the impact of AT on the curvature of the classifier decision boundary curvature and therefore on the orientation of perturbations  $\alpha$ ,  $\beta$  and  $\gamma$  for an input x.

ied thoroughly [191, 192, 188]. For example, Moosavi-Dezfooli *et al.* [191] observed high curvature of the decision boundary around most inputs after vanilla training and employed it to explain the existence of universal perturbations [189]. Furthermore, Moosavi-Dezfooli *et al.* [192] made a remarkable observation that employing single-attack AT alone reduces the curvature of the decision boundary and regularizing decision boundary curvature leads to gains in adversarial robustness.

Since single-attack AT reduces the curvature of the decision boundary, we test our hypothesis by studying the following two networks on CIFAR-10 data: a *non-robust* ResNet18  $f_{\theta}^{\text{van}}$  trained using vanilla training, and a *robust* ResNet18  $f_{\theta}^{\text{rob}}$  trained using the TRADES [22] single-attack AT framework employing  $\ell_{\infty}$  perturbations.

We compute perturbations  $\boldsymbol{\alpha}_{i}^{\kappa}$ ,  $\boldsymbol{\beta}_{i}^{\kappa}$ , and  $\boldsymbol{\gamma}_{i}^{\kappa}$  for each  $\boldsymbol{x}_{i} \in X$  for both networks, *i.e.*,  $\kappa \in \{\text{van, rob}\}$ . Now we treat them as vectors in  $\mathbb{R}^{D}$  and study their relative orientations via a subspace analysis. We compute the singular vector basis  $\mathcal{P}^{\kappa} = \{\boldsymbol{p}_{1}^{\kappa}, \ldots, \boldsymbol{p}_{D}^{\kappa}\}$  for the set of  $\ell_{2}$  bounded perturbations  $\Delta^{\kappa} = \{\boldsymbol{\beta}_{1}^{\kappa}, \ldots, \boldsymbol{\beta}_{D}^{\kappa}\} \forall i, \kappa \in \{\text{van, rob}\}$ . The singular vectors  $\boldsymbol{p}_{i}^{\kappa}$  are ordered in descending order of their singular values.

We plot the normalized mean squared projections of the three perturbation types on the perturbation basis  $\mathcal{P}^{\kappa}$  for  $\kappa \in \{\text{van, rob}\}$  in Fig. 6.4(a) and Fig. 6.4(b), respectively. The contrast is clear immediately. The perturbations of a vanilla trained network roll-off gradually to occupy a larger

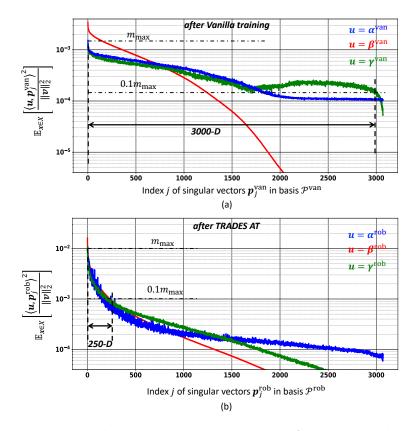


Figure 6.4: Normalized mean squared projections of three perturbation types on the singular vector basis  $\mathcal{P}^{\kappa}$  of  $\boldsymbol{\beta}^{\kappa}$  perturbations): (a) after vanilla training ( $\kappa \equiv \text{van}$ ), and (b) after TRADES training ( $\kappa \equiv \text{rob}$ ) for ResNet18 on CIFAR-10.

subspace as indicated in Fig. 6.4(a). Specifically, the projections of  $\boldsymbol{\alpha}^{\text{van}}$ and  $\boldsymbol{\gamma}^{\text{van}}$  occupy almost all 3000 directions in the basis  $\mathcal{P}^{\kappa}$ , *i.e.*, their mean squared projections are within ~ 10% of the maximum mean squared projection value  $m_{\text{max}}$ . This shows that the dominant singular vectors of  $\boldsymbol{\beta}^{\text{van}}$ are not well-aligned with  $\boldsymbol{\alpha}^{\text{van}}$  and  $\boldsymbol{\gamma}^{\text{van}}$ . This misalignment between the perturbation vectors implies that all three would need to be included during training (multi-attack AT) in order to provide robustness against the union of perturbation models. This is exactly the approach taken by Maini *et al.* [107] resulting the significant (10×) complexity increase over single-attack AT.

In contrast, with TRADES AT, all three perturbations types are squeezed into a much smaller subspace spanning only the top 250 singular vectors in the perturbation basis  $\mathcal{P}^{\text{rob}}$ . Outside these 250 dimensions, the mean squared projections fall to < 10% of their maximum value. Thus, single- $\ell_{\infty}$  attack AT enhances the alignment between different perturbations which in turn improves robustness. However, this improvement is small, e.g., while robust accuracy of 50% is achieved against  $\ell_{\infty}$  perturbations, robust accuracy of only 15% is realized against  $\ell_1$  attacks. Nevertheless, this type of subspace analysis hints at other methods to enhance the alignment between various perturbation vectors within a single-attack AT framework. The discovery of such methods will provide robustness to union of perturbation models at the complexity of single-attack AT.

In summary, the results in Fig. 6.4 validate the hypothesis that singleattack AT increases the alignment of different perturbation types *on average* due to the reduction in the decision boundary curvature around most inputs.

## 6.4 Discussion

In this chapter, we carried out subspace analysis of adversarial perturbations. Specifically, we proposed and validated two hypotheses about geometric orientations of adversarial perturbations. The SOAP hypothesis shows that the decision boundary of vanilla trained deep net gets warped around the natural images in the dimensions orthogonal to dominant image subspace. This also aligns well with the observation that deep nets overfit in the dimensions having low data variance.

The second hypothesis conveys that the different types of adversarial perturbations are squeezed into a smaller subspace even after an adversarial training employing only one type of perturbations. We exploit this insight in Chapter 7 to minimize the training cost of adversarially robust deep nets via shaped noise augmented processing.

# CHAPTER 7

# EFFICIENT AND ROBUST DEEP NET TRAINING VIA NOISE SHAPING

## 7.1 Overview

Today adversarial training (AT) provides state-of-the-art (SOTA) empirical defense against adversarial perturbations. For this, adversarial perturbations are used during training to optimize a robust loss function [13, 22, 43, 44]. Early AT frameworks [13, 22] were  $7\times$ -to- $10\times$  more computationally demanding than vanilla training. More recent works [43, 44, 47] have significantly reduced the computational demands of AT via single-step attacks and superconvergence.

However, today's AT frameworks predominantly focus on a single-attack, i.e., they seek robustness to a single perturbation, typically  $\ell_{\infty}$ -bounded [43, 44, 45, 22, 46, 47, 48, 49, 50, 51, 52, 53, 54, 193]. This results in low performance against other perturbations such as  $\ell_2$ ,  $\ell_1$ , or the union of  $(\ell_{\infty}, \ell_2, \ell_1)$ . Indeed, as shown in Fig. 7.1, four SOTA single-attack AT frameworks (black markers) employing only  $\ell_{\infty}$ -bounded perturbations achieve low adversarial accuracy  $\mathcal{A}_{adv}^{(U)}$  of  $\approx 15\%$ -to-20% against the union of  $(\ell_{\infty}, \ell_2, \ell_1)$  perturbations. Recent extensions in AT [107, 106, 194] do seek higher  $\mathcal{A}_{adv}^{(U)}$  but only at the expense of a 6×-to-10× increase in the total training time (blue markers in Fig. 7.1). The large training time of these AT frameworks has inhibited their application to large-scale datasets such as ImageNet, e.g., [107, 106] show results for MNIST and CIFAR-10 only, while [194] only additionally shows 64 × 64 ImageNet-100 results.

The high training time for AT frameworks arises from two sources: (i) the need to employ larger networks, *e.g.*, MSD [107] with ResNet-18 achieves higher  $\mathcal{A}_{adv}^{(U)}$  than PAT [194] with ResNet-50 (see Fig. 7.1); and (ii) the need to incorporate multiple perturbations during each attack step and a higher overall number of attack steps, *e.g.*, 50 in MSD [107], 20 in AVG [106].

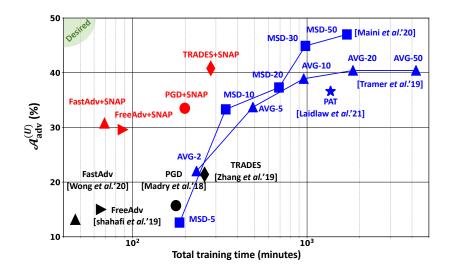


Figure 7.1: Adversarial accuracy  $(\mathcal{A}_{adv}^{(U)})$  against union of  $(\ell_{\infty}, \ell_2, \ell_1)$  vs. measured wall-clock total training time on CIFAR-10 with different AT frameworks on single NVIDIA TESLA P100 GPU.  $\epsilon = (0.031, 0.5, 12)$  for  $(\ell_{\infty}, \ell_2, \ell_1)$  perturbations, respectively. SNAP enhances robustness with a small increase in training time. All frameworks except PAT employ ResNet-18.

Obviously one can always reduce the number of attack steps in MSD/AVG to proportionally reduce training time. Doing so results in training time and  $\mathcal{A}_{adv}^{(U)}$  to rapidly approach the training complexity and  $\mathcal{A}_{adv}^{(U)}$  of standard AT frameworks, *e.g.*, a five-step MSD and two-step AVG is equivalent in training time and accuracy to PGD and TRADES, respectively. Notwithstanding the expensive nature of 50-step multi-attack training, today MSD [107] achieves a SOTA  $\mathcal{A}_{adv}^{(U)}$  of 47% with ResNet-18 on CIFAR-10.

This poses a question: Can we approach the high robustness of multipleattack AT such as 50-step MSD against the union of  $(\ell_{\infty}, \ell_2, \ell_1)$  perturbations while maintaining the low training time of fast single-attack AT frameworks such as FreeAdv [43] and FastAdv [44]?

In our quest to answer this question we find that noise augmentation using adequately shaped noise within standard single-attack AT frameworks employing  $\ell_{\infty}$ -bounded perturbations significantly improves robustness against the union of  $(\ell_{\infty}, \ell_2, \ell_1)$  perturbations. The improvement appears to be a consequence of a well-established byproduct of AT frameworks – the reduction in the curvature of the decision boundary of networks trained using singleattack AT [191, 192]. We confirm this connection by quantifying the impact of single-attack AT on the geometric orientations of different perturbations. Based on this insight, we propose Shaped Noise Augmented Processing (SNAP) - a method to enhance robustness against the union of perturbation types by augmenting single-attack AT frameworks. SNAP prepends a deep net with a shaped noise (SN) augmentation layer (see Fig. 7.2) whose distribution parameter  $\Sigma$  is learned with that of the network ( $\theta$ ) within any standard single-attack AT framework. SNAP improves the robustness of four SOTA  $\ell_{\infty}$ -AT frameworks against the union of ( $\ell_{\infty}, \ell_2, \ell_1$ ) perturbations by 15%-to-20% on CIFAR-10 (red markers in Fig. 7.1) with only a modest ( $\sim 10\%$ ) increase in training time. This expands the capabilities of widely popular single-attack  $\ell_{\infty}$  AT frameworks to providing robustness to the union of ( $\ell_{\infty}, \ell_2, \ell_1$ ) perturbations without sacrificing training efficiency. We validate SNAP's benefits via thorough comparisons with nine SOTA adversarial training and randomized smoothing frameworks across different operating regimes on both CIFAR-10 and ImageNet.

One tangible outcome of our work – we demonstrate for the first time ResNet-50 (ResNet-101) networks on ImageNet that achieve  $\mathcal{A}_{adv}^{(U)} = 32\%$ (35%) against the union of  $(\ell_{\infty}(\epsilon = 2/255), \ell_2(\epsilon = 2.0), \ell_1(\epsilon = 72.0))$  perturbations.

Our code is available at https://github.com/adpatil2/SNAP.

## 7.2 Shaped Noise Augmented Processing (SNAP)

We show that single-attack AT can be enhanced to address multiple perturbations by introducing noise to appropriately *wiggle* the  $\ell_{\infty}$ -bounded perturbations (Fig. 7.2(a)). However, to do so, the noise distribution needs to be *chosen* and *shaped* appropriately to minimize its impact on natural accuracy and robustness to  $\ell_{\infty}$ -bounded perturbations.

We experiment with both  $\ell_{\infty}$  and  $\ell_2$  perturbations in single-attack AT frameworks and find  $\ell_{\infty}$ -AT to be suitable for our proposed shaped noise augmentation (see Sec. 7.3.2 for details). Hence, in this section, we describe SNAP for single-attack AT frameworks employing  $\ell_{\infty}$  perturbations.

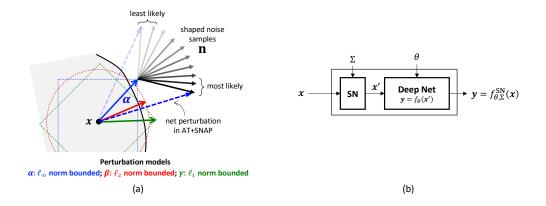


Figure 7.2: SNAP: (a) intuition underlying SNAP (not an exact depiction), and (b) SNAPnet  $f_{\theta,\Sigma}^{SN}(\boldsymbol{x})$  constructed from a given deep net  $f_{\theta}(\boldsymbol{x})$ by prepending a shaped noise (SN) augmentation layer which perturbs the primary input  $\boldsymbol{x}$  with noise  $\mathbf{n}$  whose distribution parameter  $\Sigma$  is learned during AT along with the base network parameter  $\theta$ .

#### 7.2.1 SNAPnet

A deep net  $f_{\theta}(\boldsymbol{x}) : \mathbb{R}^{D} \to \{0,1\}^{C}$  parametrized by  $\theta$  maps the input  $\boldsymbol{x} \in \mathbb{R}^{D}$  to a one-hot vector  $\boldsymbol{y} \in \{0,1\}^{C}$  over C classes.

We construct a SNAP-based deep net (SNAPnet)  $f_{\theta,\Sigma}^{\text{SN}}(\boldsymbol{x})$  by introducing an additive shaped noise (SN) layer (Fig. 7.2(b)), where the noise distribution parameter  $\Sigma$  is learned during training. Formally,

$$\boldsymbol{y} = f_{\theta,\Sigma}^{\mathrm{SN}}(\boldsymbol{x}) = f_{\theta}(\boldsymbol{x} + \mathbf{n}) = f_{\theta}(\boldsymbol{x} + V\Sigma\mathbf{n}_{0})$$
(7.1)

where  $\mathbf{n}_0 \sim \mathcal{L}(0, \mathbf{I}_{D \times D})$  is an isotropic Laplace noise vector with zero mean and identity covariance matrix,  $\Sigma = \text{Diag}[\sigma_1, \ldots, \sigma_D]$  is a distribution parameter denoting its per-dimension standard deviation,  $\mathbf{I}_{D \times D}$  denotes the  $D \times D$  identity matrix, and  $V = [\mathbf{v}_1, \ldots, \mathbf{v}_D]$  denotes a basis in  $\mathbb{R}^D$ . We also studied Gaussian and Uniform distributed  $\mathbf{n}_0$ , but empirically find the Laplace distribution to yield better results (Sec. 7.3.2). We use  $V = \mathbf{I}_{D \times D}$ for all our experiments everywhere except in Sec.7.5.2 where we study other options for V.

The final classification decision d is computed via

$$d = \arg\max_{c} \left[ \mathbb{E}_{\mathbf{n}} [\boldsymbol{y}] \right]_{c}$$
(7.2)

#### Algorithm 1 Training SNAPnet

**Input:** training set X; basis  $V = [v_1, \ldots, v_D]$ ; total noise power  $P_{\text{noise}}$ ; minibatch size r; baseline training method BASE; noise variance update frequency  $U_f$ ; Total number of epochs T**Initialize:** noise variances  $\Sigma_0 = \text{Diag}[\sigma_{1,0}, \ldots, \sigma_{D,0}]$ . **Output:** robust network  $f_{\theta,\Sigma}^{\text{SN}}$ , noise variances  $\Sigma_T = \text{Diag}[\sigma_{1,T}^2, \ldots, \sigma_{D,T}^2]$ . 1: for epoch  $t = 1 \dots T$  do for mini-batch  $B = \{x_1, \dots, x_r\}$  do  $\theta \leftarrow \text{BASE}_{\ell_{\infty}}\left(f_{\theta, \Sigma_t}^{\text{SN}}(\{x_i\}_{i=1}^r), \theta\right) \triangleright$ 2: BASE() Training end for 3: if  $t \mod U_f = 0$  then  $\triangleright$  SNAP Distribution Update once every  $U_f$ 4: epochs for mini-batch  $B = \{x_1, \ldots, x_r\}$  do 5: $\{m{x}_i^{ ext{adv}}\}_{i=1}^r \leftarrow ext{PGD}_{\ell_2}^{(K)} \Big(f_{ heta, \Sigma_t}^{ ext{SN}}ig(\{m{x}_i\}_{i=1}^rig)\Big); \quad m{\eta}_i \,=\, m{x}_i^{ ext{adv}} - m{x}_i \,\,orall \,\,i \in \mathbb{R}^{ ext{adv}}$ 6:  $\{1,\ldots,r\}$  $\gamma_j \leftarrow \gamma_j + \sum_{i=1}^r (\langle \boldsymbol{v}_j, \boldsymbol{\eta}_i \rangle)^2 \quad \forall j \in \{1, \dots, D\}$ projections; See Eq. (7.3) 7:  $\triangleright$  Accumulate end for 8:  $\sigma_{j,t+1}^2 = P_{\text{noise}} \frac{\sqrt{\gamma_j}}{\sum_{k=1}^D \sqrt{\gamma_k}} \quad \forall j \in \{1, \dots, D\} \qquad \triangleright \text{ Normalize accumulated}$ projections; See Eq. (7.3) 9: 10: else  $\Sigma_{t+1} \leftarrow \Sigma_t$ 11: 12:end if 13: end for

where  $[\boldsymbol{a}]_c$  denotes the *c*-th element of vector  $\boldsymbol{a}$ . Note, the shaped noise perturbs the input  $\boldsymbol{x}$  with a noise source  $\mathbf{n} = V\Sigma\mathbf{n}_0$  (Eq. (7.1)). The distribution parameter  $\Sigma$  is learned in the presence of any standard AT method [13, 22, 43] used for learning deep net parameters  $\theta$  as described next.

#### 7.2.2 Training SNAPnet

Algorithm 1 summarizes the procedure for training SNAPnet  $f_{\theta,\Sigma}^{SN}(\boldsymbol{x})$ . In each epoch, an arbitrary AT method BASE() (line 2) updates network parameters  $\theta$  with input perturbed by noise **n**. Here BASE() can be any established AT framework [13, 22, 43, 44] employing  $\ell_{\infty}$  perturbation.

The SNAP parameter  $\Sigma$  is updated once every  $U_f = 10$  epochs via a *SNAP* distribution update (lines 4-10). In this update, the per-dimension noise variance  $\sigma_i^2$  is updated proportional to the root mean squared projection of the adversarial perturbations  $\boldsymbol{\eta}$  on the basis V given a total noise constraint  $\sum_{j=1}^{D} \sigma_j^2 = P_{\text{noise}}$ , where  $P_{\text{noise}}$  denotes the total noise power. Formally,

$$\sigma_j^2 \propto \sqrt{\mathbb{E}_{\boldsymbol{x}\in X}(\langle \boldsymbol{\eta}, \boldsymbol{v}_j \rangle^2)}$$
 s.t.  $\sum_{j=1}^D \sigma_j^2 = P_{\text{noise}}$  (7.3)

where  $\eta$  is the  $\ell_2$  norm-bounded PGD adversarial perturbation for the given input  $\boldsymbol{x} \in X$  (line 6). Note that these  $\ell_2$  perturbations are employed only for noise shaping and are distinct from the  $\ell_{\infty}$  perturbations employed by BASE() AT (line 2). Also,  $\ell_{\infty}$  perturbations cannot be used here since their projections are constant  $\forall j$  when  $V = \mathbf{I}_{D \times D}$ , whereas employing  $\ell_1$  perturbations leads to poor shaping due to high sparsity.

Thus, in SNAP, the average squared  $\ell_2$  norm of the noise vector **n** is held constant at  $P_{\text{noise}}$  while adapting the noise variances in the individual dimensions so as to align the noise vectors with the adversarial perturbations *on average*. Intuitively, the decision boundary is pushed aggressively in those directions.

#### 7.2.3 Remarks

Note that the SNAP distribution update is distinct from BASE() AT. Hence, SNAP does not require any hyperparameter tuning in BASE(). For fairness to baselines we keep all hyperparameters identical when introducing SNAP in all our experiments. However, SNAP introduces a new hyperparameter  $P_{\text{noise}}$ , which permits to trade adversarial robustness  $\mathcal{A}_{adv}^{(U)}$  for natural accuracy  $\mathcal{A}_{nat}$ . This trade-off is explored in Sec. 7.3.3.

The computational overhead of SNAP is small (~ 10%) since the *SNAP* Distribution Update occurs once in 10 epochs using just 20% of the training data to update the noise standard deviations  $\sigma_i$ .

## 7.3 Experimental Results

#### 7.3.1 Setup

Following experimental settings of prior work [22, 43, 107], we employ a ResNet-18 network for CIFAR-10 experiments and both ResNet-50 and ResNet-101 networks for ImageNet experiments. Accuracy on clean test data is referred to with  $\mathcal{A}_{nat}$  and accuracy on adversarially perturbed test data is referred to via  $\mathcal{A}_{adv}^{(\ell_{\infty})}$ ,  $\mathcal{A}_{adv}^{(\ell_{2})}$ , and  $\mathcal{A}_{adv}^{(\ell_{1})}$ , for  $\ell_{\infty}$ ,  $\ell_{2}$ , and  $\ell_{1}$  norm bounded perturbations, respectively. Accuracy against the *union* of all three perturbations is denoted by  $\mathcal{A}_{adv}^{(U)}$ .

For a fair robustness comparison, our evaluation setup closely follows the setup of [107] for CIFAR-10 data: (1) choose norm bounds  $\epsilon = (0.031, 0.5, 12.0)$  for  $(\ell_{\infty}, \ell_2, \ell_1)$  perturbations, respectively; (2) scale norm bounds for images to lie between [0, 1]; (3) choose the PGD attack configuration to be 100 iterations with 10 random restarts for all perturbation types;<sup>1</sup> and (4) estimate  $\mathcal{A}_{adv}^{(U)}$  as the fraction of test data that is simultaneously resistant to all three perturbation models.

Following the guidelines of [14], we carefully design *adaptive* PGD attacks that target the full defense – SN layer – since SNAPnet is end-to-end differentiable. Specifically, we backpropagate to primary input  $\boldsymbol{x}$  through the SN layer (see Fig. 7.2). Thus, the final shaped noise distribution is exposed to the adversary. We also account for the expectation  $\mathbb{E}_{\mathbf{n}}[\cdot]$  in Eq. (7.2) by explicitly averaging deep net logits over  $N_0(=8)$  noise samples *before* computing the gradient, which eliminates any gradient obfuscation, and is known to be the strongest attack against noise augmented models [195].

On CIFAR-10 data, we compare with the following seven key SOTA AT frameworks: PGD [13], TRADES [22], FreeAdv [43], FastAdv [44], AVG [106], MSD [107], PAT [194]. We also compare with two randomized smoothing frameworks [196, 195]. Thanks to their GitHub code releases, we first successfully reproduce their results with a ResNet-18 network in our environment. In the case of PAT [194], we evaluate and compare with their pre-trained ResNet-50 model on CIFAR-10. We compare all training times on a single NVIDIA P100 GPU. On ImageNet data, we primarily compare to

<sup>&</sup>lt;sup>1</sup>Following [107], we also run all attacks on a subset of the first 1000 test examples with 10 random restarts for CIFAR-10 data.

Table 7.1: ResNet-18 CIFAR-10 results showing the impact of SNAP augmentation of PGD [13] AT framework with  $\ell_{\infty}$  (top) and  $\ell_2$  (bottom) perturbations where [G], [U], and [L], denote shaped Gaussian, Uniform, and Laplace noise.

Method	$\mathcal{A}_{ m nat}$	$\mathcal{A}_{\mathrm{adv}}^{(\ell_{\infty})}$ $\epsilon = 0.03$	$\mathcal{A}_{\rm adv}^{(\ell_2)}$ $\epsilon = 0.5$	$\mathcal{A}_{ m adv}^{(\ell_1)}$ $\epsilon = 12$	$\mathcal{A}_{ ext{adv}}^{(U)}$
PC	GD AT v	with $\ell_\infty$ :	perturb	ations	
PGD	84.6	48.8	62.3	15.0	15.0
+SNAP[G]	80.7	45.7	66.9	34.6	31.9
+SNAP[U]	85.1	42.7	66.7	28.6	26.6
+SNAP[L]	83.0	44.8	68.6	40.1	35.6
PC	GD AT v	with $\ell_2$ p	erturba	tions	
PGD	89.3	28.8	67.3	31.8	25.1
+SNAP[G]	83.0	35.0	65.8	39.9	30.2
+SNAP[U]	86.4	32.3	66.7	30.2	25.0
+SNAP[L]	84.8	33.4	66.1	42.5	30.8

FreeAdv [43]. We train ResNet-50 and its SNAPnet version with FreeAdv on a Google Cloud server with four NVIDIA P100 GPUs to compare their accuracy and training times. Our code and pretrained models are available at https://github.com/adpatil2/SNAP.

## 7.3.2 Impact of Noise Distribution and Model of BASE() AT Perturbations

In this subsection, we first study the impact of employing  $\ell_{\infty}$  vs.  $\ell_2$  perturbations in BASE AT() (see line 2 in Alg. 1) on  $\mathcal{A}_{adv}^{(U)}$ . For each choice, we further experiment with three distributions for the SN layer in Fig. 7.2(b) viz. Gaussian, Uniform, and Laplace. We do not consider  $\ell_1$  perturbations in BASE AT() since Maini *et al.* [107] showed that employing  $\ell_1$  single-attack AT achieves very low robustness to all attacks. We choose PGD [13] AT as BASE AT() for this ablation study. For a fair comparison across the noise distributions, we fix  $P_{\text{noise}} = 160$ , enforcing all noise vectors to have the same average  $\ell_2$  norm. For each distribution, the noise is shaped per the procedure summarized in Alg. 1.

As observed in Table 7.1,  $\ell_{\infty}$ -PGD AT achieves much lower  $\mathcal{A}_{adv}^{(U)}$  than  $\ell_{2}$ -PGD AT, an observation also reported by Maini *et al.* [107]. With SNAP, however, we find that there is an interaction between the perturbation model

Table 7.2: ResNet-18 CIFAR-10 results showing the impact of SNAP aug-
mentation of established $\ell_{\infty}$ -AT frameworks. The computational overhead
of SNAP is limited to $\sim 10\%$ .

Method	$\mathcal{A}_{ m nat}$	$\mathcal{A}_{\mathrm{adv}}^{(\ell_{\infty})}$ $\epsilon = 0.03$	$\mathcal{A}_{\rm adv}^{(\ell_2)}$ $\epsilon = 0.5$	$\mathcal{A}_{ m adv}^{(\ell_1)}$ $\epsilon = 12$	$\mathcal{A}_{ ext{adv}}^{(U)}$
High C	Complexi	ity AT w	$\mathbf{ith}\;\ell_{\infty}\;\mathbf{p}$	erturbat	ions
PGD	84.6	48.8	62.3	15.0	15.0
+SNAP	83.0	44.8	68.6	40.1	35.6
TRADES	82.1	50.2	59.6	19.8	19.7
+SNAP	80.9	45.2	66.9	46.6	41.2
Low C	omplexi	ty AT wi	ith $\ell_\infty$ pe	erturbati	ons
FreeAdv	81.7	46.1	59	15.0	15.0
+SNAP	83.5	39.7	66.2	34.3	29.6
FastAdv	85.7	46.2	60.0	13.2	13.2
+SNAP	84.2	40.4	67.9	36.6	30.8

in PGD AT and the noise distribution in SNAP. For instance, SNAP[U] enhances  $\mathcal{A}_{adv}^{(U)}$  by 11% with  $\ell_{\infty}$ -PGD AT while not achieving any improvement with  $\ell_2$ -PGD AT. In fact, SNAP appears to be particularly suitable for  $\ell_{\infty}$ -AT, since it always improves  $\mathcal{A}_{adv}^{(U)}$  by 11%-to-20.6% irrespective of the noise distribution.

Finally, of the three noise distributions, we find the Laplace distribution to be distinctly superior, achieving the highest  $\mathcal{A}_{adv}^{(U)}$  (35.6% and 30.8%) due to a significant improvement in  $\mathcal{A}_{adv}^{(\ell_1)}$  for both  $\ell_{\infty}$  and  $\ell_2$  PGD AT, respectively. The superiority of the Laplace distribution in achieving high  $\mathcal{A}_{adv}^{(\ell_1)}$  stems from its heavier tail compared to the Gaussian and Uniform distributions with the same variance. Shaped Laplace noise generates the highest fraction of extreme values in a given noise sample. Hence, it is more effective in improving accuracy against  $\ell_1$ -bounded attacks, which are the strongest when perturbing few pixels by a large magnitude [107, 106]. We discuss this further in the Sec. 7.5.3. Henceforth, unless otherwise mentioned, we choose Laplace noise for SNAP and  $\ell_{\infty}$  perturbations for BASE() AT as the default setting since it achieves the highest  $\mathcal{A}_{adv}^{(U)}$ .

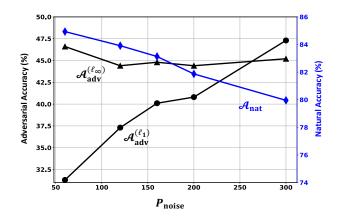


Figure 7.3: ResNet-18 CIFAR-10 results: adversarial accuracy  $\mathcal{A}_{adv}^{(\ell_1)}$ ,  $\mathcal{A}_{adv}^{(\ell_{\infty})}$ , and natural accuracy  $\mathcal{A}_{nat}$  vs. total noise power  $P_{noise}$  for PGD+SNAP.

## 7.3.3 Impact of $P_{\text{noise}}$

Next, we explore the impact of the SNAP hyperparameter  $P_{\text{noise}}$ , which constrains the average squared  $\ell_2$  norm of the noise vector **n**. It enables to trade between adversarial and natural accuracy.

Figure 7.3 shows that, as  $P_{\text{noise}}$  increases,  $\mathcal{A}_{\text{adv}}^{(\ell_1)}$  improves from 31% to 47%, accompanied by a graceful (5%) drop in  $\mathcal{A}_{\text{nat}}$  and a small drop of 2% in  $\mathcal{A}_{\text{adv}}^{(\ell_{\infty})}$  that stabilizes to  $\approx 45\%$ . These results show: (1) SNAP preserves the impact of  $\ell_{\infty}$  perturbations which is not surprising since PGD AT [13] explicitly includes those, and (2)  $P_{\text{noise}}$  provides an explicit knob to control the  $\mathcal{A}_{\text{nat}}$  vs.  $\mathcal{A}_{\text{adv}}$  trade-off. Henceforth, we choose  $P_{\text{noise}}$  values that incur < 1.5% drop in  $\mathcal{A}_{\text{nat}}$  for all SNAP+AT experiments.

#### 7.3.4 SNAP Augmented SOTA AT Frameworks

Table 7.2 shows the effectiveness of SNAP for four SOTA AT frameworks: high complexity frameworks, such as PGD [13], TRADES [22], and low complexity frameworks such as FreeAdv [43], FastAdv [44]. All are trained against  $\ell_{\infty}$  attacks with  $\epsilon = 0.031$ . As expected, while they achieve high  $\mathcal{A}_{adv}^{(\ell_{\infty})}$ , their  $\mathcal{A}_{adv}^{(\ell_{1})}$  and  $\mathcal{A}_{adv}^{(\ell_{1})}$  are lower.

For high-complexity AT, SNAP enhances  $\mathcal{A}_{adv}^{(\ell_2)}$  and  $\mathcal{A}_{adv}^{(\ell_1)}$  by ~ 6% and ~ 25%, respectively, while incurring only a drop of ~ 5% in  $\mathcal{A}_{adv}^{(\ell_\infty)}$ . Thus overall, SNAP improves robustness  $(\mathcal{A}_{adv}^{(U)})$  by ~ 20% against the *union* of

the three perturbation models. Note that this robustness improvement comes at only a ~ 1% drop in  $\mathcal{A}_{nat}$  (see Table 7.2). For low-complexity ATs, SNAP improvements in union robustness ( $\mathcal{A}_{adv}^{(U)}$ ) are also significant (~ 15%). Again, presence of SNAP improves  $\mathcal{A}_{adv}^{(\ell_2)}$  and  $\mathcal{A}_{adv}^{(\ell_1)}$ . This time the drop in  $\mathcal{A}_{adv}^{(\ell_{\infty})}$  is ~ 7%. We believe this is due to the fact that these frameworks employ weaker single-step attacks during training. Note that in the case of FreeAdv+SNAP, we actually observe a ~ 2% *increase* in  $\mathcal{A}_{nat}$ , a trend we also observe in the ImageNet experiments described later.

#### 7.3.5 Robustness vs. Training Complexity

Next we quantify adversarial robustness vs. training time trade-offs. Table 7.3 shows that SNAP augmentation of single-attack AT frameworks achieves the highest  $\mathcal{A}_{adv}^{(U)}$ , when training time is constrained to 12 hours (sets **B**, **C**, **D**, and **E**).

For instance, TRADES+SNAP achieves a 4% higher  $\mathcal{A}_{adv}^{(U)}(=41\%)$  than MSD-20 with 2 hours *lower* training time (Set **B** in Table 7.3). Similarly, PGD+SNAP achieves a 2% higher  $\mathcal{A}_{adv}^{(U)}$  than MSD-10 while having a similar training time (Set **C**). Note that both PGD and TRADES here use 100 training epochs with standard step learning rate (LR) schedule, while MSD frameworks employ a cyclic learning rate schedule to achieve superconvergence in 50 epochs.

In Set **D**, following Maini et al. [107], we employ a cyclic learning rate schedule for PGD, TRADES, as well as for PGD+SNAP and TRADES+SNAP to achieve convergence in 50 epochs. Improvements in  $\mathcal{A}_{adv}^{(U)}$  for PGD+SNAP and TRADES+SNAP are similar to those in Sets **B** and **C**. Most notably, PGD+SNAP with cyclic learning rate achieves ~ 20% and 11.5% higher  $\mathcal{A}_{adv}^{(U)}$  than MSD-5 and AVG-2, respectively, while having a similar training time (~ 3 hours). Set **E** augments the data from Table 7.2 with training times. FastAdv+SNAP and FreeAdv+SNAP achieve a high  $\mathcal{A}_{adv}^{(U)} \sim 30\%$ , while preserving the training efficiency of both FastAdv and FreeAdv. Notably, FastAdv+SNAP achieves 18% higher  $\mathcal{A}_{adv}^{(U)}$  than MSD-5, while being ~ 2.7× more efficient to train.

Table 7.3: CIFAR-10 results for comparing adversarial accuracy $\mathcal{A}_{adv}^{(U)}$ vs.
training time (on single NVIDIA P100 GPU) for different AT frameworks and
the improvements by introducing proposed SNAP technique. All frameworks
except PAT [194] (which employs ResNet-50) employ ResNet-18.

Method		Epochs	$\mathcal{A}_{\mathrm{nat}}$	$\mathcal{A}_{ ext{adv}}^{(U)}$	Total time		
	LR schedule			auv	(minutes)		
Set A: Total Time $\geq$ 12 Hrs							
AVG 50 Step [106]	cyclic	50	84.8	40.4	4217		
AVG 20 Step [106]	cyclic	50	85.6	40.4	1834		
AVG 10 Step [106]	cyclic	50	86.7	38.9	956		
PAT [194]	step	100	82.4	36.6	1364		
MSD 50 Step [107]	cyclic	50	81.7	47.0	1693		
MSD 30 Step [107]	cyclic	50	82.4	44.9	978		
Set I	B: 8 Hrs $<$ To	otal Time	e < 12	Hrs			
AVG 5 Step [106]	cyclic	50	87.8	33.7	489		
MSD 20 Step [107]	cyclic	50	83.0	37.3	690		
TRADES [22]	$\operatorname{step}$	100	82.0	19.7	516		
TRADES+SNAP	$\operatorname{step}$	100	80.9	41.2	566		
Set	C: 5 Hrs $<$ T	otal Tim	e < 8 ]	Hrs			
MSD 10 Step [107]	cyclic	50	83.6	33.3	342		
PGD [13]	$\operatorname{step}$	100	84.6	15.0	354		
PGD+SNAP	$\operatorname{step}$	100	83.0	35.6	403		
Set 1	D: 2 Hrs $<$ T	otal Tim	e < 5 ]	Hrs			
AVG 2 Step [106]	cyclic	50	88.4	22.0	232		
MSD 5 Step [107]	cyclic	50	84.0	12.6	185		
PGD [13]	cyclic	50	82.8	15.7	177		
TRADES [22]	cyclic	50	80.0	21.4	258		
PGD+SNAP	cyclic	50	82.3	33.5	199		
TRADES+SNAP	cyclic	50	78.8	40.8	280		
${\rm Set \ E: \ Total \ Time < 2 \ Hrs}$							
FreeAdv [43]	step	200	81.7	15.0	66		
FastAdv [44]	cyclic	50	85.7	13.2	47		
FreeAdv+SNAP	step	200	83.5	<b>29.6</b>	88		
FastAdv+SNAP	cyclic	50	84.2	30.8	69		

Table 7.4: ImageNet results: Iso-hyperparameter introduction of SNAP yields ~ 20% improvement in adversarial accuracy  $(\mathcal{A}_{adv}^{(U)})$  with modest impact on training time for ResNet-50 and ResNet-101.

Training	$\mathcal{A}_{ m nat}$	$\mathcal{A}_{ m adv}^{(\ell_{\infty})}$ $\epsilon = 2/255$	$\mathcal{A}_{\rm adv}^{(\ell_2)}$ $\epsilon = 2.0$	$\mathcal{A}_{\rm adv}^{(\ell_1)}$ $\epsilon = 72.0$	$\mathcal{A}_{ ext{adv}}^{(U)}$	Total time (minutes)	
	ResNet-50						
FreeAdv [43]	61.7	47.8	19.9	14.8	12.6	3590	
FreeAdv+SNAP	66.8	46.1	37.8	37.4	32.4	3756	
ResNet-101							
FreeAdv [43]	65.4	51.8	22.8	18.8	16.1	5678	
FreeAdv+SNAP	69.7	50.3	41.1	40.2	35.4	5904	

#### 7.3.6 ImageNet Results

Thanks to SNAP's low computational overhead combined with FreeAdv's fast training time, we are for the first time able to report adversarial accuracy of ResNet-50 and ResNet-101 against the union of  $(\ell_{\infty}, \ell_2, \ell_1)$  attacks on ImageNet.

We closely follow the evaluation setup of Shafahi *et al.* [43]. Specifically, we use a 100-step PGD attack, one of the strongest adversaries considered by Shafahi *et al.* [43], and evaluate on the entire test set. We first reproduce FreeAdv [43] results using the *same* hyperparameters and then introduce SNAP.

In order to clearly demonstrate the contrast between robustness to different perturbation models, we evaluate with  $\epsilon = (2/255, 2.0, 72.0)$  for  $(\ell_{\infty}, \ell_2, \ell_1)$ attacks, respectively.<sup>2</sup> As shown in Table 7.4, FreeAdv achieves a high  $\mathcal{A}_{adv}^{(\ell_{\infty})} = 47.8\%$  with ResNet-50, but a lower  $\mathcal{A}_{adv}^{(\ell_2)} = 20\%$  and  $\mathcal{A}_{adv}^{(\ell_1)} = 15\%$ , and consequently, a low  $\mathcal{A}_{adv}^{(U)}$  of 12.6% against the union of the perturbations. In contrast, FreeAdv+SNAP improves  $\mathcal{A}_{adv}^{(\ell_2)}$  and  $\mathcal{A}_{adv}^{(\ell_1)}$  by 17% and 22%, respectively, accompanied by a 5% improvement in  $\mathcal{A}_{nat}$  and a small 2% loss in  $\mathcal{A}_{adv}^{(\ell_{\infty})}$ . This results in an overall robustness improvement of 20% against the union of the perturbation models, setting a first benchmark for ResNet-50 on ImageNet. Upon increasing the network to ResNet-101, both natural and adversarial accuracies improve by  $\approx 4\%$  for FreeAdv, a trend also observed by Shafahi *et al.* [43]. SNAP further improves FreeAdv's results for  $\mathcal{A}_{nat}$  and  $\mathcal{A}_{adv}^{(U)}$  by 4.3% and 19.3%.

<sup>&</sup>lt;sup>2</sup>Note that  $\ell_2$  and  $\ell_1$  norms of PGD perturbation with  $\ell_{\infty}$  norm of 2/255 can be as large as ~ 3.0 and ~ 1100 for images of size  $224 \times 224 \times 3$ .

Table 7.5: ResNet-18 SVHN results showing the impact of SNAP augmentation of  $\ell_{\infty}$ -PGD [13] AT frameworks. Adding SNAP improves  $\mathcal{A}_{adv}^{(U)}$  by ~ 30% while having only a small impact on  $\mathcal{A}_{nat}$  and  $\mathcal{A}_{adv}^{(\ell_{\infty})}$ .

Method	$\mathcal{A}_{\mathrm{nat}}$	$\mathcal{A}_{\mathrm{adv}}^{(\ell_{\infty})}$ $\epsilon = 0.03$	$\mathcal{A}_{\rm adv}^{(\ell_2)}$ $\epsilon = 0.5$	$\mathcal{A}_{\mathrm{adv}}^{(\ell_1)}$ $\epsilon = 8$	$\mathcal{A}_{ ext{adv}}^{(U)}$
PGD	89.9	45.3	34.9	4.8	4.8
PGD+SNAP	89.3	44.0	67.4	48.3	36.3

Table 7.6: ResNet-18 CIFAR-10 results showing a comparison between MNG [197] and PGD+SNAP (from Table 7.2). All MNG numbers are exactly as reported in their paper. We reevaluate PGD+SNAP with our PGD attacks using the new  $\epsilon$  values used by [197]. PGD+SNAP achieves 3%, 2%, 4.5% higher  $\mathcal{A}_{nat}$ ,  $\mathcal{A}_{adv}^{(\ell_{\infty})}$ ,  $\mathcal{A}_{adv}^{(\ell_{1})}$ , respectively, while being at least ~ 40% faster in terms of epoch time. †: Note that MNG time is measured on NVIDIA GeForce RTX 2080Ti (by [197]), while PGD+SNAP is measured on NVIDIA Tesla P100. An RTX 2080Ti has 20% more CUDA cores than a Tesla P100.

Method	$\mathcal{A}_{ m nat}$	$\mathcal{A}_{\mathrm{adv}}^{(\ell_{\infty})}$ $\epsilon = 0.03$	$\mathcal{A}_{\mathrm{adv}}^{(\ell_2)}$ $\epsilon = 0.31$	$\mathcal{A}_{ m adv}^{(\ell_1)} \ \epsilon=8$	Time per Epoch (seconds)
MNG [197]	79.8	43.9	75.8	53.8	$354^{\dagger}$
PGD+SNAP	83.1	45.9	74.1	58.3	240

#### 7.3.7 SVHN Results

Table 7.5 shows PGD and PGD+SNAP results on SVHN data. We train both PGD and PGD+SNAP models for 100 epochs using a piece-wise LR schedule. We start with an initial LR of 0.01 and decay it once at the 95th epoch.

In Table 7.5, we observe a trend that is similar to our observations for CIFAR-10 and ImageNet results. In particular, for SVHN, SNAP turns out to be even more effective, with ~ 30% improvement in  $\mathcal{A}_{adv}^{(U)}$  while almost preserving both  $\mathcal{A}_{nat}$  and  $\mathcal{A}_{adv}^{(\ell_{\infty})}$ .

### 7.3.8 Comparison with Madaan *et al.* [197]

The meta-noise generator (MNG) [197] employs a multi-layer deep net to generate noise samples during AT. Importantly, MNG still employs multiple attacks during training, but samples only one of the attacks randomly at a time to reduce the training cost.

Table 7.7: ResNet-18 CIFAR-10 results showing SNAP's impact on the prediction complexity, where  $N_0$  denotes the number of noise samples employed to estimate  $\mathbb{E}[\cdot]$  in Eq. (7.2). We find that for mere accuracy estimation, even a single forward pass ( $N_0 = 1$ ) suffices.  $\pm xx$  denotes the standard deviation over 10 independent test runs.

Method	$\mathcal{A}_{\mathrm{nat}}$ (%)
TRADES	81.7
TRADE	S+SNAP
$N_0 = 1$	$80.1 \pm 0.22$
$N_0 = 2$	$80.3 \pm 0.14$
$N_0 = 4$	$80.7 {\pm} 0.12$
$N_0 = 8$	$80.9 {\pm} 0.10$
$N_0 = 16$	$80.9 {\pm} 0.08$

However, they have yet to release their code or pretrained models even though their work was posted on arXiv a year ago. Absence of public codes from Madaan *et al.* [197] makes it difficult to clearly compare with their work, especially in terms of training time. Nonetheless, in this section, our goal is to ensure that the comparison is fair. Table 7.6 reports natural and adversarial accuracy of MNG against ( $\ell_{\infty}, \ell_2, \ell_1$ ) attacks as reported by Madaan *et al.* [197]. We find that PGD+SNAP achieves 3%, 2%, 4.5% higher  $\mathcal{A}_{nat}$ ,  $\mathcal{A}_{adv}^{(\ell_{\infty})}$ , and  $\mathcal{A}_{adv}^{(\ell_1)}$ , respectively. Note that Madaan *et al.* [197] evaluate  $\mathcal{A}_{adv}^{(\ell_{\infty})}$ and  $\mathcal{A}_{adv}^{(\ell_2)}$  against PGD-50 attacks, whereas here we employ PGD-100 attacks and, following their protocol, evaluate on the entire CIFAR-10 dataset with a single restart. Furthermore, epoch time for PGD+SNAP is 1.4× smaller than that of MNG [197] even though MNG time was measured on a more recent NVIDIA RTX 2080Ti, which has 20% more CUDA cores than the Tesla P100 GPU that we used for PGD+SNAP.

Importantly, a key advantage of SNAP is its scalability. We are able to report robust ResNet-50 and ResNet-101 networks on ImageNet (Table 7.4), whereas Madaan *et al.* [197] report results only up to  $64 \times 64$  TinyImageNet.

## 7.3.9 Impact of SNAP on Prediction Complexity

While SNAP augmentation has a modest impact on the training time (Table 7.3), here we check whether it could *potentially* increase the model prediction complexity due to the need to estimate the expectation  $\mathbb{E}[\cdot]$  in Eq. (7.2). As expected, by increasing  $N_0$ , the deviation of the  $\mathcal{A}_{nat}$  estimate reduces (see Table 7.7). However, we find that for accuracy estimation, a single forward pass ( $N_0 = 1$ ) suffices. Specifically, an  $\mathcal{A}_{nat}$  estimate with  $N_0 = 1$ is within 1% of the  $\mathcal{A}_{nat}$  estimate with  $N_0 = 16$ . Furthermore, even with  $N_0 = 1$ , the standard deviation of  $\mathcal{A}_{nat}$  is as low as ~ 0.2%. Thus, the impact of SNAP on prediction complexity can be very small.

## 7.4 Robustness Stress Tests

We conduct robustness stress tests to confirm that the benefits of SNAP are sustained for a range of attack norm-bounds, larger number of attack steps, and even for "gradient-free" attacks. For these experiments, we consider networks trained using TRADES and TRADES+SNAP (rows in Table 7.2), since they achieve the highest  $\mathcal{A}_{adv}^{(U)}$  among the four SOTA AT frameworks.

### 7.4.1 Sweeping Norm-bounds and Number of Attack Steps

We sweep the number of PGD attack steps (K) and norm-bounds  $(\epsilon)$  for all three perturbations  $(\ell_{\infty}, \ell_2, \ell_1)$  to confirm that the robustness gains from SNAP are achieved for a wider range of attack norm bounds, and are sustained even after increasing attack steps.

Figure 7.4(a)-(c) validates the Table 7.2 conclusion that TRADES+SNAP achieves large gains (~ 20%) in  $\mathcal{A}_{adv}^{(\ell_1)}$  and  $\mathcal{A}_{adv}^{(\ell_2)}$  with a small (~ 4%) drop in  $\mathcal{A}_{adv}^{(\ell_{\infty})}$ . Furthermore, this conclusion holds for a large range of  $\epsilon$  values for all three perturbations. Additionally, the gain in  $\mathcal{A}_{adv}^{(\ell_2)}$  due to SNAP at  $\epsilon = 1.2$ is greater than the one reported in Table 7.2 for  $\epsilon = 0.5$ .

Now we increase the attack steps K to 500 and observe the impact on adversarial accuracy against  $(\ell_{\infty}, \ell_2, \ell_1)$  perturbations in Fig. 7.4(d,e,f), respectively. In all cases, we observe hardly any change of the adversarial accuracy beyond K = 100. Hence, we have chosen K = 100 for all our experiments.

Recall we employ 10 random restarts as recommended by [107] for *all* our adversarial accuracy evaluations on CIFAR-10 data.

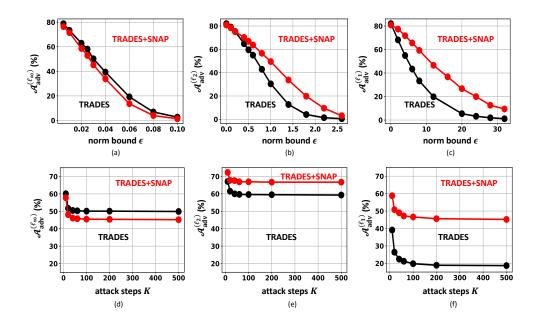


Figure 7.4: ResNet-18 CIFAR-10 results: Adversarial accuracy vs. norm bound  $\epsilon$  for: (a)  $\ell_{\infty}$ , (b)  $\ell_2$ , (c)  $\ell_1$  PGD-100 attack. Adversarial accuracy vs. attack steps K for (d)  $\ell_{\infty}$  ( $\epsilon = 0.031$ ), (e)  $\ell_2$  ( $\epsilon = 0.5$ ), (f)  $\ell_1$  ( $\epsilon = 12$ ) PGD-100 attacks.

#### 7.4.2 Evaluating Robustness Against New Attacks

We evaluate adversarial accuracy against the recent DDN [31], Boundary [32], and Square [33] attacks. The DDN attack was shown to be one of the SOTA gradient-based attacks, while boundary attack is one of the strongest "gradient-free" attacks. Of all the attacks considered in [107], PGD turns out to be the strongest for  $\ell_{\infty}$  and  $\ell_1$  perturbations. Hence, in this section, we evaluate against  $\ell_2$  norm-bounded DDN, boundary, and Square attacks.

Following [107], we use the FoolBox [198] implementation of the boundary attack, which uses 25 trials per iteration. For the DDN attack, we use 100 attack steps with appropriate logit averaging for  $N_0 = 8$  noise samples *before* computing the gradient in each step (similar to our PGD attack implementations). As mentioned in the Sec. 7.3.1, it eliminates any gradient obfuscation due to the presence of noise.

Table 7.8 shows that SNAP improves adversarial accuracy against the DDN attack by ~ 6%. This is similar to improvements seen against  $\ell_2$ -PGD attack in Table 7.2. Similarly, TRADES+SNAP achieves 3.5% (4.5%) higher adversarial accuracy than TRADES against the Boundary [32] (Square [33])

Table 7.8: ResNet-18 CIFAR-10 results showing natural accuracy (%) and adversarial accuracy (%) against  $\ell_2$  norm bounded DDN attack [31], boundary attack [32], and Square [33] for TRADES and TRADES+SNAP networks from Table 7.2.

	TRADES	TRADES+SNAP
Natural Accuracy	82.1	80.9
DDN [31] ( $\epsilon = 0.5$ )	59.7	65.8
Boundary [32] ( $\epsilon = 0.5$ )	63.5	67.0
Square [33] ( $\epsilon = 0.5$ )	68.2	72.7

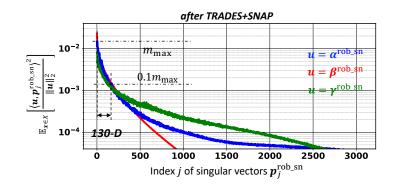


Figure 7.5: Normalized mean squared projections of three perturbation types on the singular vector basis  $\mathcal{P}^{\kappa}$  of  $\ell_2$  perturbations of ResNet18 on CIFAR-10 after TRADES+SNAP training ( $\kappa \equiv \text{rob}_{\text{sn}}$ ). The singular vectors  $\boldsymbol{p}_i^{\kappa}$  comprising  $\mathcal{P}^{\kappa} = \{\boldsymbol{p}_1^{\kappa}, \ldots, \boldsymbol{p}_D^{\kappa}\}$  are ordered in descending order of their singular values.

attack.

## 7.5 Additional Investigations

## 7.5.1 Subspace Analysis of Adversarial Perturbations for TRADES+SNAP Model

In this section, we carry out a subspace analysis of adversarial perturbations (similar to Sec. 6.3 in Chapter 6) for TRADES+SNAP. We confirm that our hypothesis in Sec. 6.3 holds even after SNAP augmentation of TRADES. Following the same experimental setup and the notation from Sec. 6.3, we compute perturbations  $\alpha_i$ ,  $\beta_i$ , and  $\gamma_i$  for each  $x_i \in X$  for ResNet-18 trained using TRADES+SNAP, *i.e.*,  $\kappa \equiv \text{rob\_sn}$ . We compute the singular vector

Table 7.9: ResNet-18 CIFAR-10 results showing the impact of noise shaping basis V for  $\ell_{\infty}$ -PGD [13] AT framework with SNAP. In this table, SNAP[G], SNAP[U], and SNAP[L] denote shaped noise augmentations with Gaussian, Uniform, and Laplace noise distributions, respectively, and  $U_{\text{img}}$  refers to the singular vector basis of the training images.

Method	$\mathcal{A}_{ m nat}$	$\mathcal{A}_{\mathrm{adv}}^{(\ell_{\infty})}$ $\epsilon = 0.03$	$\mathcal{A}_{\rm adv}^{(\ell_2)}$ $\epsilon = 0.5$	$\mathcal{A}_{ m adv}^{(\ell_1)}$ $\epsilon=12$	$\mathcal{A}_{ ext{adv}}^{(U)}$		
PGD	84.6	48.8	62.3	15.0	15.0		
N	Noise shaping basis $V = \mathbf{I}_{D \times D}$						
+SNAP[G]	80.7	45.7	66.9	34.6	31.9		
+SNAP[U]	85.1	42.7	66.7	28.6	26.6		
+SNAP[L]	83.0	44.8	68.6	40.1	35.6		
N	Noise shaping basis $V = U_{img}$						
+SNAP[G]	81.7	48.9	67.5	29.8	28.7		
+SNAP[U]	82.0	46.6	67.8	27.8	25.7		
+SNAP[L]	81.7	46.8	65.9	28.5	27.4		

basis  $\mathcal{P}^{\kappa}$  for the set of  $\ell_2$  bounded perturbations  $\Delta^{\kappa} = \{\beta_1^{\kappa}, \ldots, \beta_{|X|}^{\kappa}\}$ . Figure 7.5 plots the normalized mean squared projections of the three types of perturbation vectors on the singular vector basis  $\mathcal{P}^{\kappa}$  of a TRADES+SNAP trained ResNet-18. We find that the projections generally follow the same trend as those for a TRADES-trained network which are shown in Fig. 6.4(b). However, we also notice that after SNAP augmentation, the three perturbation types get squeezed into an even smaller 130-dimensional subspace, *i.e.*, projections are < 10% of the maximum projection value for all dimensions beyond the first 130 dimensions.

#### 7.5.2 Impact of Noise Shaping in the Image Basis

Recall that, for all experiments till now, we chose the noise shaping basis  $V = \mathbf{I}_{D \times D}$ , *i.e.*, the noise was shaped and added in the standard basis in  $\mathbb{R}^D$ , where  $\mathbf{I}_{D \times D}$  denotes the identity matrix (see Eq. (7.1) and Algorithm 1).

In this section, we explore the shaped noise augmentation in the *image* basis, *i.e.*, singular vector basis of the training set images. Specifically, we choose  $V = U_{img} = [\mathbf{u}_1, \ldots, \mathbf{u}_D]$ , where  $U_{img}$  denotes the singular vector basis of the images in the training set. Thus, the sampled noise vector  $\mathbf{n}_0$ (see Eq. (7.1)) is scaled by direction-wise standard deviation matrix  $\Sigma$  and rotated by  $U_{\text{img}}$  before being added to the input image  $\boldsymbol{x}$ .

The rationale for choosing  $V = U_{\text{img}}$  is as follows: Recent works [190, 43, 199] have demonstrated the generative behavior of adversarial perturbations of networks trained with single-attack AT, *i.e.*, adversarial perturbations of robust networks exhibit semantics similar to the input images. Thus, the perturbation basis of the robust networks trained with single-attack AT seems to be aligned with the image basis (see Chapter 6).

We repeat the experiments in Table 7.1 while keeping all the settings *identical* except for choosing  $V = U_{img}$  instead of  $V = \mathbf{I}_{D \times D}$ . Table 7.9 shows the results. The first three rows correspond to  $V = \mathbf{I}_{D \times D}$  and are reproduced from Table 7.1. Note that, in order to preserve  $\mathcal{A}_{nat} > 81\%$ , we need to reduce  $P_{noise} = 60$  when  $V = U_{img}$ , since the noise is now pixel-wise correlated.

In Table 7.9, we notice that  $\mathcal{A}_{adv}^{(\ell_1)}$  is significantly reduced when  $V = U_{img}$  as compared to the case  $V = \mathbf{I}_{D \times D}$ . More interestingly, all three types of noise distributions result in similar values for  $\mathcal{A}_{adv}^{(\ell_1)}$  when  $V = U_{img}$ . We discuss this phenomenon in the next section, *i.e.*, Sec. 7.5.3.

Table 7.9 shows that the orientation of a noise vector is as important as its distribution. The simpler choice of  $V = \mathbf{I}_{D \times D}$  turns out to be more effective.

## 7.5.3 Understanding the effectiveness of SNAP[L] for $\ell_{\infty}$ AT

In this subsection, we conduct additional studies to further understand the following two observations in SNAP: (i) shaped Laplace noise is particularly effective (Table 7.1), and (ii) rotating noise vectors ( $V = U_{img}$ ) reduces their effectiveness (Table 7.9). We study the properties of the noise vector **n** for different noise distributions.

We conjecture that the Laplace distribution is most effective because of its heavier tail compared to Gaussian and Uniform distributions of the same variance. A long-tailed distribution will generate more large magnitude elements in a vector drawn from it and hence is more effective in emulating a strong  $\ell_1$ -norm bounded perturbation. Furthermore, the standard (unrotated) basis preserves this unique attribute of samples drawn from such distributions.

This conjecture is validated by Fig. 7.6(a) which shows that noise samples drawn from the Laplace distribution in the standard basis have the highest

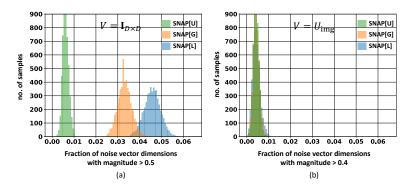


Figure 7.6: ResNet18 CIFAR-10 results: histograms of the fraction of noise vector dimensions with magnitude (a) > 0.5 when  $V = \mathbf{I}_{D \times D}$ , and (b) > 0.4 when  $V = U_{\text{img}}$ . Histograms are plotted for 5000 random noise samples **n**. The three shaped noise distributions are from the corresponding networks in Table 7.9.

average number of dimensions with large (> 0.5) magnitudes, followed by Gaussian and Uniform distributions. This correlates well with the results in Table 7.1 and Table 7.9 (first three rows), in that  $\mathcal{A}_{adv}^{(\ell_1)}$  is the highest for Laplace followed by those for Gaussian and Uniform. Additionally, the use of  $V = U_{img}$  dissolves this distinction between the three distributions as shown in Fig. 7.6(b) which explains the similar (and lower)  $\mathcal{A}_{adv}^{(\ell_1)}$  values for all three distributions in Table 7.9.

Thus, we confirm that the type of noise plays an important role in robustifying single-attack  $\ell_{\infty}$  AT frameworks to the union of multiple perturbation models. Specifically, the noise vectors with higher fraction of noise dimensions with larger magnitudes are better at complementing  $\ell_{\infty}$  AT frameworks.

#### 7.5.4 Evaluating Common Corruptions and Functional Attack

In this section, we check if there are any other downsides of SNAP when it improves robustness against the union of  $(\ell_{\infty}, \ell_2, \ell_1)$  perturbations. In particular, we check if SNAP improvements are achieved at the cost of a drop in accuracy against common corruptions [10] or functional adversarial attacks [200].

We use corrupted images provided by [10] to estimate accuracy in the presence of common corruptions ( $\mathcal{A}_{cc}$ ). We average the accuracy numbers across different corruption strengths and types. Also, we use the ReColorAdv

Table 7.10: ResNet-18 CIFAR-10 results showing natural accuracy  $\mathcal{A}_{nat}$ , adversarial accuracy  $\mathcal{A}_{adv}^{(U)}$  against the union of  $(\ell_{\infty}, \ell_2, \ell_1)$  perturbations, accuracy  $\mathcal{A}_{cc}$  in the presence of common corruptions [10], and adversarial accuracy  $\mathcal{A}_{adv}^{(f)}$  against a functional adversarial attack ReColorAdv [200]. All accuracy numbers are in %. In this table,  $\mathbf{I}_{D\times D}$  denotes *D*-dimensional identity matrix, while  $U_{img}$  denotes singular vector basis of the training images. We find that SNAP augmentations of  $\ell_{\infty}$ -PGD significantly ( $\approx 20\%$ ) improve  $\mathcal{A}_{adv}^{(U)}$  while preserving both  $\mathcal{A}_{cc}$  and  $\mathcal{A}_{adv}^{(f)}$ .

Method	$\mathcal{A}_{ ext{nat}}$	$\mathcal{A}_{ ext{adv}}^{(U)}$	$\mathcal{A}_{ m cc}$	$\mathcal{A}_{\mathrm{adv}}^{(\mathrm{f})}$ ReColorAdv		
Vanilla	94.5	0.0	72.0	0.9		
$\ell_{\infty}$ -PGD	84.6	15.0	75.6	53.5		
N	Noise shaping basis $V = \mathbf{I}_{D \times D}$					
+SNAP[G]	80.7	31.9	72.8	55.1		
+SNAP[U]	85.1	26.6	75.0	46.9		
+SNAP[L]	83.0	35.6	75.3	51.3		
Noise shaping basis $V = U_{img}$						
+SNAP[G]	81.7	28.7	73.6	54.5		
+SNAP[U]	82.0	25.7	73.1	54.0		
+SNAP[L]	81.7	27.4	73.4	55.3		

setup of [194] to estimate accuracy against functional adversarial attacks  $(\mathcal{A}_{adv}^{(f)})$ . We also make it *adaptive* to our defense framework via appropriate noise averaging (similar to our adaptive PGD attacks [195]) to eliminate any gradient obfuscations. As observed in Table 7.10, SNAP augmentations of PGD AT generally preserve both  $\mathcal{A}_{cc}$  and  $\mathcal{A}_{adv}^{(f)}$ . In particular, 20.6% improvement in  $\mathcal{A}_{adv}^{(U)}$  via PGD+SNAP[L] (with  $V = \mathbf{I}_{D \times D}$ ) is accompanied with the same  $\mathcal{A}_{cc}$  and only a 2.2% lower  $\mathcal{A}_{adv}^{(f)}$  (= 51.3%) compared to PGD AT. In contrast, vanilla training achieves an  $\mathcal{A}_{adv}^{(f)}$  of only 0.9%. Even with  $V = U_{img}$ , PGD+SNAP[L] achieves a 1.8% higher  $\mathcal{A}_{adv}^{(f)}$  along with a 12.4% improvement in  $\mathcal{A}_{adv}^{(U)}$ . Note that all  $\mathcal{A}_{adv}^{(U)}$  numbers are idential to the ones reported in Sec. 7.5.2.

We conclude that SNAP augmentation of PGD AT improves  $\mathcal{A}_{adv}^{(U)}$  by up to 20% while preserving its robustness against common corruptions and functional adversarial attacks. Thus, SNAP expands the capabilities of  $\ell_{\infty}$  AT frameworks without any significant downside. However, further work is required to improve robustness to a larger class adversarial attacks, such as rotation [201], texture [202], etc., simultaneously.

## 7.6 Relationship between SNAP and Randomized Smoothing

In this section, we discuss a relationship between SNAP and Randomized Smoothing (RS) [196, 195], another popular noise augmentation approach that employs isotropic Gaussian noise in order to obtain certification bounds on adversarial robustness. Specifically, we show that SNAP[G] is closely related to RS, *i.e.*, SNAP[G] can be viewed as RS with shaped Gaussian noise. In Sec. 7.6.1, we first extend the generalize the bound derived in [196] (for isotropic Gaussian noise) and use it to justify our noise shaping approach in Eq. (7.3). We further provide geometric intuition (in Sec. 7.6.2) demonstrating the why shaped noise inherently leads to improved robustness compared to isotropic noise for Gaussian distribution. At the end, in Sec. 7.6.3, we demonstrate that the noise shaping in SNAP enables significantly better  $\mathcal{A}_{nat}$  vs.  $\mathcal{A}_{adv}^{(U)}$  trade-off compared due to RS.

#### 7.6.1 Theoretical Justification for Noise Shaping

We define a smoothed classifier [196] as follows:

**Definition 7.1.** Given a classifier  $f : \mathbb{R}^D \to \mathcal{Y}$ , for a given input  $\boldsymbol{x}$ , a smoothed classifier g is defined as:

$$g(\boldsymbol{x}) = \arg \max_{c \in \mathcal{Y}} \Pr\{f(\boldsymbol{x} + \mathbf{n}) = c\}$$

where  $\mathbf{n} \sim \mathcal{N}(0, C_{\sigma}), C_{\sigma} \in \mathbb{R}^{D \times D}$  denotes the covariance matrix, and  $\mathcal{Y}$  denotes the set of discrete class labels.

Notice that SNAPnet decision rule (see Eq. (7.2)) is same as smoothed classifier above when (i) SNAP employs Gaussian distributed noise (SNAP[G]), and (ii)  $C_{\sigma} = \text{Diag}[\sigma_1, \ldots, \sigma_D]$ , where dimensionwise noise standard deviation are shaped according to Eq. (7.3). In [196], a certification bound for a smoothed classifier was obtained for the isotropic case, *i.e.*, when  $C_{\sigma} = \sigma^2 \mathbf{I}_{D \times D}$ . Here, we first generalize this bound for the anisotropic case where  $C_{\sigma}$  is an arbitrary covariance matrix as stated in Theorem 7.1. We then employ Theorem 7.1 to obtain a theoretical justification for noise shaping (as per Eq. (7.3)) with Gaussian noise distribution. **Theorem 7.1.** Let  $f : \mathbb{R}^D \to \mathcal{Y}$  be an arbitrary function either deterministic or random,  $\mathbf{n} \sim \mathcal{N}(0, C_{\sigma})$  with a covariance matrix  $C_{\sigma} \in \mathbb{R}^{D \times D}$ , and  $g(\mathbf{x}) = \arg \max_c \Pr(f(\mathbf{x} + \mathbf{n}) = c)$ . For a specific  $\mathbf{x} \in \mathbb{R}^D$ , if  $\exists c_A \in \mathcal{Y}$  and  $\rho_A, \rho_B \in [0, 1]$  such that:

$$\Pr\{f(\boldsymbol{x} + \mathbf{n}) = c_A\} \ge \rho_A \ge \rho_B \ge \max_{c \neq c_A} \Pr\{f(\boldsymbol{x} + \mathbf{n}) = c\}$$
(7.4)

then, for any  $\boldsymbol{\xi} \in \mathbb{R}^D$ , we obtain  $g(\boldsymbol{x} + \boldsymbol{\xi}) = c_A$  if

$$\sqrt{\boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{\xi}} < \frac{\boldsymbol{\Phi}^{-1}(\rho_A) - \boldsymbol{\Phi}^{-1}(\rho_B)}{2}$$
(7.5)

where  $\mathbf{\Phi}$  denotes the cumulative density function of a standard 1-D Gaussian distribution  $\mathcal{N}(0,1)$ .

*Proof.* The detailed proof provided in Appendix C follows the sequence of arguments provided in [196] for the isotropic case.  $\Box$ 

Since Theorem 7.1 holds for an arbitrary Gaussian distribution  $\mathbf{n} \sim \mathcal{N}(0, C_{\sigma})$ , we treat  $C_{\sigma}$  as a design variable and ask the following question: what is a good choice for  $C_{\sigma}$  in order to enhance robustness under a fixed total power  $P_{noise}$  constraint? Equation (7.5) provides a hint. It suggests that an ideal choice for  $C_{\sigma}$  is one that minimizes the LHS of Eq. (7.5) when  $\boldsymbol{\xi} = \boldsymbol{\eta}$ , *i.e.*,  $\boldsymbol{\xi}$ is an adversarial perturbation, *e.g.*, a PGD adversarial perturbation. More formally, we would like to choose  $C_{\sigma}$  as a solution to the following constrained optimization problem:

$$C_{\sigma}^{*} = \arg\min_{C_{\sigma}} \mathbb{E}_{\boldsymbol{x} \in X} \left[ \boldsymbol{\eta}^{T} C_{\sigma}^{-1} \boldsymbol{\eta} \right] \quad \text{s.t.} \quad \text{Tr}[C_{\sigma}] = P_{\text{noise}}$$
(7.6)

where  $\operatorname{Tr}[C_{\sigma}]$  denotes the trace of matrix  $C_{\sigma}$ , and  $\eta$  being the PGD adversarial perturbation is a function of the input  $\boldsymbol{x} \in X$ .

Without loss of generality, we decompose  $C_{\sigma} = U\Sigma U^T$  using an arbitrary orthonormal basis  $U = [\mathbf{u}_1, \ldots, \mathbf{u}_D]$  of  $\mathbb{R}^D$ , where  $\Sigma = \text{Diag}[\sigma_1^2, \ldots, \sigma_D^2]$ . Note:  $P_{\text{noise}} = \sum_{j=1}^D \sigma_j^2 = \text{Tr}[\Sigma]$ . Employing this decomposition for  $C_{\sigma}$ , we express the objective function in Eq. (7.6) as:

$$\mathbb{E}_{\boldsymbol{x}\in X}\left[\boldsymbol{\eta}^{T}C_{\sigma}^{-1}\boldsymbol{\eta}\right] = \mathbb{E}_{\boldsymbol{x}\in X}\left[\boldsymbol{\eta}^{T}U\Sigma^{-1}U^{T}\boldsymbol{\eta}\right] = \sum_{j=1}^{D}\frac{\mathbb{E}_{\boldsymbol{x}\in X}\left(\langle\boldsymbol{\eta},\boldsymbol{u}_{j}\rangle^{2}\right)}{\sigma_{j}^{2}} \qquad (7.7)$$

For a specific orthonormal basis U, the optimal  $\Sigma_U = \text{Diag}[\sigma_{u_1}^2, \ldots, \sigma_{u_D}^2]$  can be obtained as follows:

$$\sigma_{\boldsymbol{u}_j}^2 \propto \sqrt{\mathbb{E}_{\boldsymbol{x}\in X}(\langle \boldsymbol{\eta}, \boldsymbol{u}_j \rangle^2)} \quad \forall \quad j \in \{1, \dots, D\} \quad \text{s.t.} \quad \sum_{j=1}^D \sigma_{\boldsymbol{u}_j}^2 = P_{\text{noise}} \quad (7.8)$$

which is indeed the exact noise variance allocation rule in SNAP (see lines 6-10 in Alg. 1 and Eq. (7.3)). Note that this rule suggests assigning variance in proportion to the root mean-squared projections of  $\eta$  in the directions  $u_j$ . While the optimality of this rule is derived here only for Gaussian noise distribution, we find that this noise shaping rule also works well with other (Laplace, Uniform) distributions (as demonstrated in Sec. 7.3). In Sec. 7.6.2, we provide a geometric intuition for the role of such noise shaping in achieving higher robustness.

## 7.6.2 Geometric Intuition: Spherical vs. Shaped Gaussian Noise

In this section, we provide a geometric intuition (see Fig. 7.7) to demonstrate the effectiveness of noise shaping. For simplicity, we consider Gaussian noise augmented two-dimensional binary linear classifier for this analysis. We show that the classifier with SNAP[G] forces the decision boundary to be farther away from the input in order to achieve the same decision confidence compared to any other shape for augmented Gaussian noise.

Consider three binary (classes  $c_A$  and  $c_B$ ) linear classifiers  $f^{\alpha} : \mathbb{R}^2 \to \{c_A, c_B\}$  in Fig. 7.7, where  $\alpha \in \{\text{wht, wrst, sn}\}$  identifies the noise distribution type under a fixed power constraint  $P_{\text{noise}}(\sigma_{1(\alpha)}^2 + \sigma_{2(\alpha)}^2)$  defined as follows ( $\hat{\boldsymbol{\eta}}$  is a unit vector in the dominant adversarial perturbation direction, and  $\hat{\boldsymbol{\eta}}^{\perp}$  is its orthogonal direction):

- 1.  $\alpha = \text{wht}$  (Fig. 7.7(a)): The classifier is augmented with white Gaussian noise  $\mathbf{n}_{(\text{wht})}$  such that  $P_{\text{noise}} = 2\sigma^2$  and  $\sigma_{1(\text{wht})}^2 = \sigma_{2(\text{wht})}^2 = \sigma^2$ .
- 2.  $\alpha = \text{wrst}$  (Fig. 7.7(b)): The classifier is augmented with shaped noise  $\mathbf{n}_{(\text{wrst})}$  such that the noise variance  $\sigma_{1(\text{wrst})}^2$  ( $\sigma_{2(\text{wrst})}^2$ ) along  $\hat{\boldsymbol{\eta}}$  ( $\hat{\boldsymbol{\eta}}^{\perp}$ ) is decreased (increased) and  $P_{\text{noise}} = \sigma_{1(\text{wrst})}^2 + \sigma_{2(\text{wrst})}^2 = 2\sigma^2$ . Note that this variance allocation is the opposite (worst) to the one in Eq. (7.3).

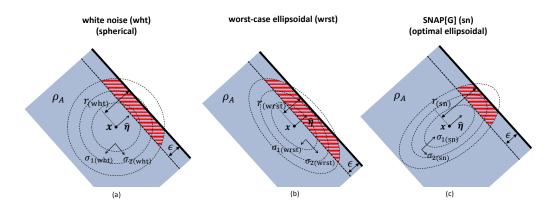


Figure 7.7: Geometric intuition regarding the impact of noise shaping on the robustness of binary linear classifiers  $f^{\alpha}$  ( $\alpha \in \{\text{wht, wrst, sn}\}$ ) augmented with noise  $\mathbf{n}_{(\alpha)}$  of a fixed power  $P_{\text{noise}} = (\sigma_{1(\alpha)}^2 + \sigma_{2(\alpha)}^2)$  when subject to the adversarial perturbation  $\boldsymbol{\eta} = \epsilon \hat{\boldsymbol{\eta}}$  where  $\hat{\boldsymbol{\eta}}$  is the unit vector in direction of the dominant adversarial perturbation for input  $\boldsymbol{x}$ : (a) white noise ( $\alpha = \text{wht}$ ) with  $\sigma_{1(\text{wht})} = \sigma_{2(\text{wht})}$ , (b) worst-case ellipsoidal noise ( $\alpha = \text{wrst}$ ) with  $\sigma_{1(\text{wrst})} < \sigma_{2(\text{wrst})}$ , and (c) SNAP[G] ( $\alpha = \text{sn}$ ) where  $\sigma_{1(\text{sn})} > \sigma_{2(\text{sn})}$ . Since the confidence  $\Pr\{f^{\alpha}(\boldsymbol{x} + \mathbf{n}_{(\alpha)}) = c_A\} = \rho_{A(\alpha)} = \rho_A, \forall \alpha \in \{\text{wht, wrst, sn}\}$  and  $\sigma_{1(\text{sn})} > \sigma_{1(\text{wrst})} > \sigma_{1(\text{wrst})}$ , it can be shown that the distance r from  $\boldsymbol{x}$  to the decision boundary satisfies  $r_{(\text{sn})} > r_{(\text{wrst})}$ . Furthermore, it can also be shown that the confidence  $\hat{\rho}_{A(\alpha)}$  in the presence of an adversarial input  $\boldsymbol{x} + \epsilon \hat{\boldsymbol{\eta}}$  satisfies  $\hat{\rho}_{A(\text{sn})} > \hat{\rho}_{A(\text{wht})} > \hat{\rho}_{A(\text{wrst})}$  as can be intuited from the striped red area.

3.  $\alpha = \text{en}$  (Fig. 7.7(c)): The classifier is augmented with shaped (Gaussian) noise  $\mathbf{n}_{(\text{sn})}$  such that the noise variance  $\sigma_{1(\text{sn})}^2 (\sigma_{2(\text{sn})}^2)$  along  $\hat{\boldsymbol{\eta}} (\hat{\boldsymbol{\eta}}^{\perp})$  is *increased* (*decreased*) and  $P_{\text{noise}} = \sigma_{1(\text{sn})}^2 + \sigma_{2(\text{sn})}^2 = 2\sigma^2$ . Note that this variance allocation is the one prescribed SNAP in Eq. (7.3).

If  $(\boldsymbol{x}, c_A)$  is an (input, class label) pair and all three classifiers  $f^{\alpha}$  achieve the same confidence level  $\rho_A$  in classifying  $\boldsymbol{x}$  correctly, *i.e.*,

$$\Pr\{f^{\alpha}(\boldsymbol{x} + \mathbf{n}_{(\alpha)}) = c_A\} = \rho_A \quad \forall \quad \alpha \in \{\text{wht, wrst, sn}\}$$
(7.9)

then  $\rho_A$  corresponds to the mass of the probability density function of  $\mathbf{n}^{\alpha}$ on the side of the decision boundary corresponding to the class label  $c_A$  as indicated in grey in Fig. 7.7. Therefore, the distance  $r_{(\alpha)}$  of  $\boldsymbol{x}$  to the decision boundary of  $f^{\alpha}$  is given by  $r_{(\alpha)} = \sigma_{1(\alpha)} \Phi^{-1}(\rho_A)$  where  $\sigma_{1(\alpha)}^2$  is the variance of noise in the direction of  $\hat{\boldsymbol{\eta}}$ . Since  $\sigma_{1(\text{en})} > \sigma_{1(\text{wht})} > \sigma_{1(\text{wrst})}$ , therefore  $r_{\text{sn}} > r_{\text{wht}} > r_{\text{wrst}}$ , *i.e.*, SNAP[G] pushes the decision region further away as compared to any other shaped noise augmentation, including white noise. Additionally, it can be shown that the confidence  $\hat{\rho}_{A(\alpha)}$  in the presence of an adversarial input  $\boldsymbol{x} + \epsilon \hat{\boldsymbol{\eta}}$  is given by:

$$\widehat{\rho}_{A(\alpha)} = \Pr\{f^{\alpha}(\boldsymbol{x} + \epsilon \boldsymbol{\eta} + \mathbf{n}_{(\alpha)}) = c_A\} = \Phi\left(\Phi^{-1}(\rho_A) - \frac{\epsilon}{\sigma_{1(\alpha)}}\right)$$
(7.10)

and since  $\sigma_{1(\text{sn})} > \sigma_{1(\text{wht})} > \sigma_{1(\text{wrst})}$  it follows that  $\hat{\rho}_{A(\text{sn})} > \hat{\rho}_{A(\text{wht})} > \hat{\rho}_{A(\text{wrst})}$ . The striped red area in Fig. 7.7 provides a geometric intuition underlying this result. This shows that SNAP[G] provides the highest confidence for the adversarially perturbed input  $\boldsymbol{x} + \epsilon \hat{\boldsymbol{\eta}}$  as compared to any other shaped Gaussian noise augmentation including white noise. Recall that all three classifiers  $f^{\alpha}$ classify clean input  $\boldsymbol{x}$  with equal confidence of  $\rho_A$ . Thus, the *reduction* in the classifier's confidence  $(\rho_A - \hat{\rho}_{A(\alpha)})$  due to adversarial perturbation  $\epsilon \hat{\boldsymbol{\eta}}$  is the *minimum* for the SNAP[G] classifier, compared to a classifier with any other Gaussian noise augmentation. This indicates the superior robustness of SNAP[G].

#### 7.6.3 Quantitative Comparison

In this subsection, we compare with two SOTA RS frameworks, namely, RandSmooth [196], and SmoothAdv [195]. They employ isotropic Gaussian noise. In Fig. 7.8(a), we find that PGD+SNAP[L] achieves a better  $\mathcal{A}_{nat}$  vs.  $\mathcal{A}_{adv}^{(U)}$  trade-off compared to both RandSmooth [196], and SmoothAdv [195]. Specifically, note that SmoothAdv [195] can also be viewed as isotropic Gaussian augmentation of  $\ell_2$ -PGD AT. Importantly, PGD+SNAP[L] achieves a 12% higher  $\mathcal{A}_{adv}^{(U)}$  for the same  $\mathcal{A}_{nat}$ . This demonstrates the efficacy of *shaped noise* in SNAP, which enhances the robustness to the union of ( $\ell_{\infty}, \ell_2, \ell_1$ ) perturbations.

In order to further quantify importance of *noise shaping*, we also compare  $\ell_{\infty}$ -PGD+SNAP[L] with  $\ell_{\infty}$ -PGD+Iso[L], a stronger baseline alternative consisting of *isotropic* Laplace noise augmentation, *i.e.*, without any noise shaping. Specifically, in Iso[L], the noise standard deviation is *identical* in each direction, *i.e.*,  $\Sigma = \text{Diag}\left[\sqrt{\frac{P_{\text{noise}}}{D}}, \ldots, \sqrt{\frac{P_{\text{noise}}}{D}}\right]$ . Note that such distributions have recently been explored for RS [203].

Figure 7.8(b) plots the  $\mathcal{A}_{nat}$  vs.  $\mathcal{A}_{adv}^{(U)}$  trade-off for PGD+SNAP (*red* 

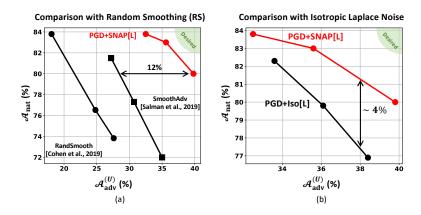


Figure 7.8: ResNet-18 CIFAR-10 results: (a)  $\mathcal{A}_{nat}$  vs.  $\mathcal{A}_{adv}^{(U)}$  for RandSmooth [196], SmoothAdv [195], and PGD+SNAP[L]; (b)  $\mathcal{A}_{nat}$  vs.  $\mathcal{A}_{adv}^{(U)}$ for PGD+SNAP[L] and PGD+Iso[L], where Iso[L] denotes a baseline SNAP alternative employing isotropic Laplace noise augmentation, *i.e.*, without noise shaping. PGD+SNAP[L] achieves better  $\mathcal{A}_{nat}$  vs.  $\mathcal{A}_{adv}^{(U)}$  trade-off due to noise shaping.

curve) and PGD+Iso[L] (black curve) by sweeping  $P_{\text{noise}}$ . We find that PGD+SNAP[L] achieves a better  $\mathcal{A}_{\text{nat}}$  vs.  $\mathcal{A}_{\text{adv}}^{(U)}$  trade-off compared to PGD+ Iso[L] by making more efficient use of noise power via noise shaping. Specifically, for  $\mathcal{A}_{\text{adv}}^{(U)} \approx 38$ , PGD+SNAP[L] achieves a ~ 4% higher  $\mathcal{A}_{\text{nat}}$ .

## 7.7 Related Works

We categorize works on adversarial robustness of DNNs into following three categories:

Low-complexity Adversarial Training: The high computational needs of AT frameworks has spurred significant efforts in reducing their complexity [47, 43, 44, 46]. FreeAdv [43] updates weights while accumulating multiple attack iterations. FastAdv [44] employs *appropriate* use of single-step attacks, while Zheng *et al.* [46] leverage inter-epoch similarity between adversarial perturbations. However, these fast AT methods seek robustness against a single perturbation type, *e.g.*,  $\ell_{\infty}$  norm-bounded perturbations. In contrast, SNAP expands the capabilities of these AT frameworks by enhancing robustness to the union of three perturbation types ( $\ell_{\infty}, \ell_2, \ell_1$ ), while preserving their efficiency.

Robustness against Union of Perturbation Models: The focus on the

robustness against the union of multiple perturbation types is relatively new. Kang et al. [204] studied transferability between different perturbation types, while Jordan et al. [205] considered combination attacks with low perceptual distortion. Stutz et al. [206] proposed a modification in AT to detect images with different models of perturbations via confidence thresholding, but they do not attempt to *classify* perturbed images correctly. For accurate classification in the presence of different perturbation models, Tramer *et al.* [106]studied empirical and theoretical trade-offs involved in including multiple perturbation types simultaneously during training. Maini et al. [107] further built upon this work to propose the multi steepest descent (MSD) AT framework which chooses one among the three perturbation models  $(\ell_{\infty}, \ell_2, \ell_1)$  in each attack iteration during training, achieving SOTA adversarial accuracy on CIFAR-10 against the union of the  $(\ell_{\infty}, \ell_2, \ell_1)$  perturbation models, albeit at a high  $(10\times)$  training time. In contrast, SNAP provides high robustness against the union of  $(\ell_{\infty}, \ell_2, \ell_1)$  perturbation models using established singleattack  $\ell_{\infty}$  AT frameworks. This enables to showcase the benefits of our approach on large-scale datasets such as ImageNet.

Recently, Laidlaw *et al.* [194] developed a novel AT framework (PAT) with low perceptual distortion attacks to demonstrate impressive generalization to unseen attacks. In contrast, we focus on extending the capabilities of widely popular  $\ell_{\infty}$ -AT frameworks to providing robustness against the union of  $(\ell_{\infty}, \ell_2, \ell_1)$  perturbations, while preserving their training efficiency.

Noise Augmentation: Multiple recent works have investigated the role of randomization in enhancing adversarial robustness [99, 207, 208, 209] with theoretical guarantees. Another prominent line of work in this category is randomized smoothing [196, 195, 210, 203], where random noise is used as a tool to compute certification bounds. Rusak *et al.* [211] also explored the role of noise augmentation for improving the robustness against common-corruptions [10]. In contrast, in SNAP, noise augmentation is used as a means to enable widely popular  $\ell_{\infty}$ -AT frameworks to efficiently achieve high robustness against the union of multiple norm-bounded perturbations. As is the characteristic of AT works, our results are primarily empirical in nature. Hence, we follow recent guidelines [14, 107] to evaluate the accuracy against the strongest possible adversaries. We do explicitly compare  $\ell_{\infty}$ -AT+SNAP with randomized smoothing approaches in the Sec. 7.6.3.

## 7.8 Discussion

In this chapter, we demonstrate how shaped noise can be introduced in deep nets to enhance their adversarial robustness vs. cost trade-off. Specifically, we propose SNAP as an augmentation that generalizes the effectiveness of  $\ell_{\infty}$ -AT to the union of  $(\ell_{\infty}, \ell_2, \ell_1)$  perturbations. SNAP's strength is its simplicity and efficiency. Consequently, this work sets a first benchmark for ResNet-50 and ResNet-101 networks which are resilient to the union of  $(\ell_{\infty}, \ell_2, \ell_1)$ perturbations on ImageNet. Note that norm-bounded perturbations include a large class of attacks, *e.g.*, gradient-based [13, 31, 106, 107, 30, 11], decisionbased [32] and black-box [33] attacks.

More work is needed to extend the proposed SNAP technique to attacks beyond norm-bounded additive perturbations, *e.g.*, functional [200, 212], rotation [201], texture [202], etc. It is important to note that SNAP is meant to be an efficient technique for improving  $\ell_{\infty}$ -AT, and *not* a new defense. Indeed defending against a large variety of attacks simultaneously remains an open problem, with encouraging results from recent efforts [107, 194].

Another limitation of our approach is that its benefits are demonstrated empirically. It is an inevitable consequence of a lack of any theoretical guarantees for underlying AT frameworks. An interesting direction of future work is to explore whether any theoretical guarantees can be derived for anisotropic shaped noise distributions in SNAP by building upon the recent developments in randomized smoothing [195, 203]. This could be a potential avenue for bridging the gap between certification bounds and empirical adversarial accuracy.

# CHAPTER 8

# CONCLUSION AND FUTURE WORK

The robustness and cost challenges of deep nets discussed in Chapter 1 are well-recognized today. In response, a large number of works have independently focused on either the accuracy vs. robustness trade-off, or the accuracy vs. cost trade-off. However, as we discussed in Chapter 1, these two trade-offs start getting interwoven as one strives to push the limits of either robustness or cost. Hence, there is a need to explicitly focus on the robustness vs. cost trade-off. However it remains little explored today. In this dissertation, we attempt to fill this gap by developing techniques to improve the robustness vs. cost trade-off in the presence of both hardware noise and data noise.

## 8.1 Summary of Contributions

In this dissertation, we studied and improved the robustness vs. cost tradeoff in diverse problem settings, ranging from the error-resilient beyond-CMOS spintronic implementations to the efficient training of robust deep nets. Specifically, we showed that both data noise and hardware noise can be shaped to improve robustness vs. cost trade-off. For example, in spintronic hardware implementations, we showed that the device-level trade-off between switching error rate, energy, and delay can be exploited to shape the error distribution at the system-level. We achieved  $1000 \times$  higher tolerance to the device error rate and  $3 \times$  smaller energy per decision via proposed efficient statistical error compensation. Furthermore, we also demonstrated an improvement in the robustness vs. cost trade-off via a synergistic combination of a novel spin-based analog circuit design and system-level classifier ensemble techniques. The resulting spin channel networks achieve up to  $10 \times$  higher energy-efficiency compared to CMOS digital implementations. At the other end of the design stack, we proposed a shaped noise augmentation technique to achieve high adversarial robustness for deep nets, while maintaining their training cost comparable to the standard vanilla training. Here we demonstrated that the noise can be used as a feature and its effectiveness can be enhanced by shaping its distribution. We achieved the best adversarial robustness vs. training cost trade-off compared to the nine state-of-the-art (SOTA) related works. Specifically, we achieved  $4\times$ smaller training time while maintaining the same robustness compared to the SOTA approach published just last year. Furthermore, We analyzed SNR of in-memory computing (IMC) employing resistive crossbar memory arrays. We identified SNR optimal array and sensing circuit configurations by understanding their impact on quantization vs. clipping noise trade-off. We considered both voltage and current driven crossbars, and provided design guidelines based on our analysis.

## 8.2 Future Prospects

Based on our work in this dissertation, we identify following four directions for promising future work:

Theoretical Guarantees on Robustness Enhancements via Noise **Shaping**: The enhancements in the robustness vs. cost trade-offs via noise shaping (see e.q. Chapter 2 and Chapter 7) demonstrated in this dissertation, while significant, are primarily empirical in nature. We believe one can derive theoretical guarantees for such robustness improvements via noise shaping. For example, in Sec. 7.6, we already discussed the relation between our proposed noise shaping techniques and randomized smoothing (RS), a promising approach for providing theoretical guarantees on adversarial robustness of deep nets [196]. We also provide the analysis for binary linear classifier showing the efficacy of the shaped noise in achieving better robustness than the isotropic noise. While the early work on RS focused on isotropic Gaussian noise [196, 195, 213, 210], more recently a larger variety of noise distributions and randomization techniques are explored [203, 214, 215, 216, 217]. These extensions can be leveraged to derive theoretical guarantees on robustness in the presence of noise shaping. This can be one of the promising ways to bridge the gap between empirical robust accuracy and certified robust

accuracy.

Noise Shaping and Error Compensation for In-Memory Computing (IMC): We discussed in Chapter 4 and Chapter 5 how computation SNR in resistive crossbar based in-memory computing implementations suffers due to the increased vulnerability to conductance and input DAC variations. We further observed that this challenge is particularly exacerbated in MRAM due to its lower conductance values. Since the root cause of all these variations is the fundamental device-level unreliability, we believe one needs to employ system-level noise shaping and error compensation techniques to improve the SNR. For resistive crossbar based IMCs, such schemes would inevitably involve some sort of on-chip chip-specific adaptive mechanism. For example, recent work [21] proposed adaptive write mechanism to compensate the write-time conductance variations in RRAM-based IMC. Similarly, even Joshi et al. [8] employ on-chip adaptive activation normalization to compensate for the resistance drifts in PCM devices. While these are encouraging initial works, we believe there is a much broader scope to employ error compensation techniques [4, 5, 7] for in-memory computing architectures.

Exploiting Hardware Noise for Enhancing Adversarial Robust**ness**: In this dissertation, while we consider the impact of both hardware and data noise, we do so in the separate problem setups. In addition to our findings in Chapter 7, recent theoretical work [207, 209] has underscored the importance of randomness in enhancing adversarial robustness of deep nets. As discussed in Chapters 2-5, emerging logic and memory devices (e.q. spintronic, RRAM, and others) are inherently unreliable. So, an interesting avenue to explore is whether their hardware noise can be exploited to enhance adversarial robustness of deep nets. An ideal place to start this exploration is the deep net inference implementations employing IMC architectures, where its analog computation provides an additional source of hardware noise. Indeed, the recent work by Cherupally et al. [218] exploring the impact of IMC noise on adversarial robustness is a good first step. However, they only evaluate against weaker black-box attacks. Hence, much more work and a rigorous evaluation against the strongest white-box attacks needs to occur in order to develop techniques for exploiting hardware noise for adversarial robustness.

Finally, we would like to note that in all our contributions improving the robustness vs. cost trade-off in this dissertation, we did incur small drop in the accuracy. It seemed inevitable since any enhancement in the adversarial robustness that sustained the scrutiny of the strongest attacks was always accompanied with a drop in the natural accuracy (recall, e.q. AT works [22, 43, 44, 45, 46, 47, 48, 49, 50, 51, 52, 53, 54]). In the research community, the working hypothesis has been that such robustness may inherently be at the odds with the accuracy [55]. Furthermore, the robustness and efficiency of ML systems are still nowhere near human capabilities despite the plethora of works have been focusing on improving them in the past few years. For instance, a human brain performs incredibly complex inference tasks accurately, efficiently (consuming  $\sim$  tens of Watts of power), and is largely robust to visual irregularities/corruptions such as fog, abnormal positions, irregular patches, and others. Thus, achieving human-level accuracy, robustness, and efficiency in the inference tasks remains to be a widely open problem even today. The recent efforts have been encouraging. For instance, demonstrated efficacy of adversarial training in learning robust features [190, 78, 219], possibility of recovering the accuracy drop in AT via large unlabeled data [220], and initial attempts at thinking about accuracy, robustness, and efficiency simultaneously [193, 221]. However, much more work needs to occur in this direction.

# APPENDIX A DERIVATION OF EQ. (5.4)

We apply the Kirchhoff's current law at the *i*th SL in Fig. 5.2 to obtain:

$$I_{\rm SL,i} = \sum_{j=1}^{N} \frac{(V_j - R_s I_{\rm SL,i})}{R_{i,j}}$$
(A.1)

where the bitcell resistance  $R_{i,j} = \frac{1}{G_{i,j}}$ . After substituting the value of  $R_{i,j}$  and rearranging the terms, we get,

$$I_{\text{SL},i} = \sum_{j=1}^{N} V_j G_{i,j} - R_s I_{\text{SL},i} \left[ \sum_{j=1}^{N} G_{i,j} \right]$$
(A.2)

Substituting  $\left[\sum_{j=1}^{N} G_{i,j}\right] = \frac{1}{R_{\text{arr},i}}$ , we get,

$$I_{{\rm SL},i} = \sum_{j=1}^{N} V_j G_{i,j} - \frac{R_s}{R_{{\rm arr},i}} I_{{\rm SL},i}$$
(A.3)

$$\left[\frac{R_{\operatorname{arr},i} + R_s}{R_{\operatorname{arr},i}}\right] I_{\operatorname{SL},i} = \sum_{j=1}^N V_j G_{i,j} \tag{A.4}$$

$$I_{\mathrm{SL},i} = \left[\frac{R_{\mathrm{arr},i}}{R_{\mathrm{arr},i} + R_s}\right] \sum_{j=1}^N V_j G_{i,j} \tag{A.5}$$

Now we introduce the VDAC constraint,  $V_{2k-1} = -V_{2k}$   $k \in \{1, \ldots, \frac{N}{2}\}$ , discussed in Sec. 5.3 to obtain

$$I_{\mathrm{SL},i} = \left[\frac{R_{\mathrm{arr},i}}{R_{\mathrm{arr},i} + R_s}\right] \sum_{k=1}^{\frac{N}{2}} V_{2k} \Delta G_{i,2k} \tag{A.6}$$

where  $\Delta G_{i,2k} = G_{i,2k-1} - G_{i,2k}$ . Note that in the absence of any variations or noise,  $I_{\text{SL},i} = I_{\text{sig},i}$  (see Eq. (5.1)). Thus, we get,

$$I_{\text{sig},i} = \left[\frac{R_{\text{arr},i}}{R_{\text{arr},i} + R_s}\right] \sum_{k=1}^{\frac{N}{2}} V_{2k} \Delta G_{i,2k} \tag{A.7}$$

which is same as the Eq. (5.4).

# APPENDIX B DERIVATION OF EQ. (5.12)

Figure B.1 shows current driven  $2 \times N$  crossbar. Based on charge conservation principle, we know that,

$$I_{\rm SL,1} + I_{\rm SL,2} = \sum_{i=1}^{N} I_i$$
 (B.1)

Let  $I_{BC,i,j}$  denote the current though a bitcell in *i*th row and *j*th column. By applying Kirchhoff's voltage law at the *j*th BL in Fig. B.1, we get,

$$I_{\text{SL},1}R_s + I_{\text{BC},1,j}R_{1,j} = I_{\text{SL},2}R_s + I_{\text{BC},2,j}R_{2,j}$$
(B.2)

Note that from Kirchhoff's current law at the *j*th BL, we have  $I_{BC,2,j} = I_j - I_{BC,1,j}$ . Substituting, we get,

$$I_{\text{SL},1}R_s + I_{\text{BC},1,j}R_{1,j} = I_{\text{SL},2}R_s + (I_j - I_{\text{BC},1,j})R_{2,j}$$
(B.3)

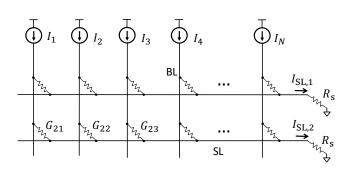


Figure B.1: Current driven  $2 \times N$  crossbar.

Rearranging the terms, we get,

$$I_{\text{BC},1,j} = \frac{\left[I_{\text{SL},2} - I_{\text{SL},1}\right]R_s}{R_{1,j} + R_{2,j}} + \frac{R_{2,j}I_j}{R_{1,j} + R_{2,j}}$$
(B.4)

Substituting  $R_{i,j} = \frac{1}{G_{i,j}}$  in the second term, we get,

$$I_{\text{BC},1,j} = \frac{\left[I_{\text{SL},2} - I_{\text{SL},1}\right]R_s}{R_{1,j} + R_{2,j}} + \frac{G_{1,j}I_j}{G_{1,j} + G_{2,j}}$$
(B.5)

Now, we apply Kirchhoff's current law at the 1st SL to obtain

$$I_{\rm SL,1} = \sum_{j=1}^{N} I_{\rm BC,1,j}$$
 (B.6)

We substitute the expression for  $I_{BC,1,j}$  from Eq. (B.5) and rearrange the terms to obtain

$$I_{\rm SL,1} = R_s \left[ I_{\rm SL,2} - I_{\rm SL,1} \right] \left[ \sum_{j=1}^N \frac{1}{R_{1,j} + R_{2,j}} \right] + \sum_{j=1}^N \frac{G_{1,j} I_j}{G_{1,j} + G_{2,j}}$$
(B.7)

Let

$$\frac{1}{R_{\rm arr}} \triangleq \left[\sum_{j=1}^{N} \frac{1}{R_{1,j} + R_{2,j}}\right] \tag{B.8}$$

Substituting and rearranging the terms, we get,

$$I_{\rm SL,1}\left[1 + \frac{R_s}{R_{\rm arr}}\right] = \frac{R_s}{R_{\rm arr}}I_{\rm SL,2} + \sum_{j=1}^N \frac{G_{1,j}I_j}{G_{1,j} + G_{2,j}}$$
(B.9)

Now we have two equations, namely Eq. (B.1) and Eq. (B.9), to solve for the two unknowns  $I_{\text{SL},1}$  and  $I_{\text{SL},2}$ . Substituting  $I_{\text{SL},2} = \left(\sum_{j=1}^{N} I_j\right) - I_{\text{SL},1}$  in Eq. (B.9) and rearranging the terms, we get,

$$I_{\rm SL,1}\left[1 + \frac{2R_s}{R_{\rm arr}}\right] = \frac{R_s}{R_{\rm arr}}\left[\sum_{j=1}^N I_j\right] + \sum_{j=1}^N \frac{G_{1,j}I_j}{G_{1,j} + G_{2,j}}$$
(B.10)

Further rearranging the terms, we get,

$$I_{\rm SL,1} = \frac{R_s}{R_{\rm arr} + 2R_s} \left[ \sum_{j=1}^N I_j \right] + \frac{R_{\rm arr}}{R_{\rm arr} + 2R_s} \left[ \sum_{j=1}^N \frac{G_{1,j}I_j}{G_{1,j} + G_{2,j}} \right]$$
(B.11)

Similarly,

$$I_{\rm SL,2} = \frac{R_s}{R_{\rm arr} + 2R_s} \left[ \sum_{j=1}^N I_j \right] + \frac{R_{\rm arr}}{R_{\rm arr} + 2R_s} \left[ \sum_{j=1}^N \frac{G_{2,j}I_j}{G_{1,j} + G_{2,j}} \right]$$
(B.12)

We know that in the absence of any variations or noise  $I_{\text{SL},i} = I_{\text{sig},i}$ , giving us the Eq. (5.12).

# APPENDIX C PROOF OF THEOREM 7.1

In this appendix, we prove Theorem 7.1. As noted earlier, this theorem is a general version of the certification bound proved in [196] for isotropic Gaussian noise. Hence, the proof here broadly follows a similar outline as that of [196]. The main distinction is due to the relaxation of isotropy of the distribution. We provide the complete proof for the sake of completeness.

## C.1 Preliminary Lemmas

We begin by restating the following Neyman-Pearson lemma:

**Lemma C.1** (Neyman-Pearson [222, 196]). Let  $\mathbf{x}$  and  $\mathbf{y}$  be random variables in  $\mathbb{R}^D$  with densities  $\mu_{\mathbf{x}}$  and  $\mu_{\mathbf{y}}$ . Let  $h : \mathbb{R}^D \to \{0, 1\}$  be a random or deterministic function. Then:

- 1. If  $S = \left\{ \boldsymbol{z} \in \mathbb{R}^D : \frac{\mu_{\mathbf{y}}(\boldsymbol{z})}{\mu_{\mathbf{x}}(\boldsymbol{z})} \leq t \right\}$  for some t > 0 and  $\mathbb{P}(h(\mathbf{x}) = 1) \geq \mathbb{P}(\mathbf{x} \in S)$ , then  $\mathbb{P}(h(\mathbf{y}) = 1) \geq \mathbb{P}(\mathbf{y} \in S)$ .
- 2. If  $S = \left\{ \boldsymbol{z} \in \mathbb{R}^D : \frac{\mu_{\mathbf{y}}(\boldsymbol{z})}{\mu_{\mathbf{x}}(\boldsymbol{z})} \ge t \right\}$  for some t > 0 and  $\mathbb{P}(h(\mathbf{x}) = 1) \le \mathbb{P}(\mathbf{x} \in S)$ , then  $\mathbb{P}(h(\mathbf{y}) = 1) \le \mathbb{P}(\mathbf{y} \in S)$ .

*Proof.* See [196].

Now, we prove the following theorem for non-isotropic Gaussian distributions with different means, as a special case of the above lemma.

Lemma C.2 (Neyman-Pearson lemma for Gaussians with different means). Let  $\mathbf{x} \sim \mathcal{N}(\mathbf{x}, C_{\sigma})$  and  $\mathbf{y} \sim \mathcal{N}(\mathbf{x} + \boldsymbol{\xi}, C_{\sigma})$ , where  $C_{\sigma} \in \mathbb{R}^{D \times D}$ denotes the covariance matrix. Let  $h : \mathbb{R}^{D} \to \{0, 1\}$  be any deterministic or random function. Then:

1. If 
$$S = \left\{ \boldsymbol{z} \in \mathbb{R}^{D} : \boldsymbol{\xi}^{T} C_{\sigma} \boldsymbol{z} \leq \beta \right\}$$
 for some  $\beta$  and  $\mathbb{P}(h(\mathbf{x}) = 1) \geq \mathbb{P}(\mathbf{x} \in S)$ ,  
then  $\mathbb{P}(h(\mathbf{y}) = 1) \geq \mathbb{P}(\mathbf{y} \in S)$ .  
2. If  $S = \left\{ \boldsymbol{z} \in \mathbb{R}^{D} : \boldsymbol{\xi}^{T} C_{\sigma} \boldsymbol{z} \geq \beta \right\}$  for some  $\beta$  and  $\mathbb{P}(h(\mathbf{x}) = 1) \leq \mathbb{P}(\mathbf{x} \in S)$ ,  
then  $\mathbb{P}(h(\mathbf{y}) = 1) \leq \mathbb{P}(\mathbf{y} \in S)$ .

*Proof.* This lemma is a straightforward application of Lemma C.1 when  $\mu_{\mathbf{x}}$  and  $\mu_{\mathbf{y}}$  are Gaussians with different means. Specifically, the ratio of densities  $\frac{\mu_{\mathbf{y}}(\mathbf{z})}{\mu_{\mathbf{x}}(\mathbf{z})}$  reduces to a linear equation  $\boldsymbol{\xi}^T C_{\sigma} \mathbf{z}$  for any given  $\mathbf{z}$  when the densities are Gaussians with different means.

It can be shown as follows:

$$\frac{\mu_{\mathbf{y}}(\boldsymbol{z})}{\mu_{\mathbf{x}}(\boldsymbol{z})} = \frac{\exp\left\{-\frac{1}{2}\left[\boldsymbol{z} - (\boldsymbol{x} + \boldsymbol{\xi})\right]C_{\sigma}^{-1}\left[\boldsymbol{z} - (\boldsymbol{x} + \boldsymbol{\xi})\right]\right\}}{\exp\left\{-\frac{1}{2}\left[\boldsymbol{z} - \boldsymbol{x}\right]C_{\sigma}^{-1}\left[\boldsymbol{z} - \boldsymbol{x}\right]\right\}}$$
(C.1)  
$$= \exp\left[\frac{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{z} + \boldsymbol{z}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}{2} - \left(\frac{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{x} + \boldsymbol{x}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}{2} + \frac{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}{2}\right)\right]$$
(C.2)

Note that  $\boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{z} = (\boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{z})^T = \boldsymbol{z}^T C_{\sigma}^{-1} \boldsymbol{\xi}$ , since  $\boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{z}$  is a scalar quantity and  $C_{\sigma}^{-1}$  is a symmetric matrix. Hence, substituting, we obtain

$$\frac{\mu_{\mathbf{y}}(\boldsymbol{z})}{\mu_{\mathbf{x}}(\boldsymbol{z})} = \exp\left[\boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{z} - \boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{x} - \frac{\boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{\xi}}{2}\right]$$
(C.3)

For any given t > 0, we can choose  $\beta = \ln t + \boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{x} + \frac{\boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{\xi}}{2}$ , such that

$$\frac{\mu_{\mathbf{y}}(\boldsymbol{z})}{\mu_{\mathbf{x}}(\boldsymbol{z})} \le t \Leftrightarrow \boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{z} \le \beta$$
(C.4)

$$\frac{\mu_{\mathbf{y}}(\boldsymbol{z})}{\mu_{\mathbf{x}}(\boldsymbol{z})} \ge t \Leftrightarrow \boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{z} \ge \beta$$
(C.5)

which concludes the proof.

## C.2 Theorem Proof

Now, we restate the Theorem 7.1 and provide its proof.

**Theorem 7.1.** Let  $f : \mathbb{R}^D \to \mathcal{Y}$  be an arbitrary function either deterministic or random,  $\mathbf{n} \sim \mathcal{N}(0, C_{\sigma})$  with a covariance matrix  $C_{\sigma} \in \mathbb{R}^{D \times D}$ , and  $g(\boldsymbol{x}) = \arg \max_c \Pr(f(\boldsymbol{x} + \mathbf{n}) = c)$ . For a specific  $\boldsymbol{x} \in \mathbb{R}^d$ , if  $\exists c_A \in \mathcal{Y}$  and  $\rho_A, \rho_B \in [0, 1]$  such that:

$$\Pr\{f(\boldsymbol{x} + \mathbf{n}) = c_A\} \ge \rho_A \ge \rho_B \ge \max_{c \neq c_A} \Pr\{f(\boldsymbol{x} + \mathbf{n}) = c\}$$
(C.6)

then, for any  $\boldsymbol{\xi} \in \mathbb{R}^d$ , we obtain  $g(\boldsymbol{x} + \boldsymbol{\xi}) = c_A$  if

$$\sqrt{\boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{\xi}} < \frac{\boldsymbol{\Phi}^{-1}(\rho_A) - \boldsymbol{\Phi}^{-1}(\rho_B)}{2}$$
(C.7)

where  $\Phi$  denotes the cumulative density function of a standard 1-D Gaussian distribution  $\mathcal{N}(0,1)$ .

*Proof.* From the definition of  $g(\boldsymbol{x})$ , we know that  $g(\boldsymbol{x} + \boldsymbol{\xi}) = c_A$ , iff

$$\Pr[f(\boldsymbol{x} + \boldsymbol{\xi} + \mathbf{n}) = c_A] > \max_{c \neq c_A} \Pr(f(\boldsymbol{x} + \boldsymbol{\xi} + \mathbf{n}) = c)$$
(C.8)

Thus, we need to show that  $\Pr[f(\boldsymbol{x} + \boldsymbol{\xi} + \mathbf{n}) = c_A] > \Pr(f(\boldsymbol{x} + \boldsymbol{\xi} + \mathbf{n}) = c)$  for every class  $c \neq c_A$ . Without loss of generality, let us carry out the analysis for one such class  $c_B$ .

We start by defining the following two simplifying notations:

$$\mathbf{x}_{\text{nat}} = \boldsymbol{x} + \mathbf{n} \tag{C.9}$$

$$\mathbf{x}_{\text{per}} = \boldsymbol{x} + \boldsymbol{\xi} + \mathbf{n} \tag{C.10}$$

Note that  $\mathbf{x}_{nat}$  and  $\mathbf{x}_{per}$  are random vectors due to randomness of  $\mathbf{n}$ , whereas  $\boldsymbol{x}$  and  $\boldsymbol{\xi}$  are deterministic vectors.

Using this notation and from Eq. (7.4), we know that

$$\Pr[f(\mathbf{x}_{nat})] \ge \rho_A$$
 and  $\Pr[f(\mathbf{x}_{nat})] \le \rho_B$  (C.11)

Our goal is to show that

$$\Pr[f(\mathbf{x}_{\text{per}}) = c_A] > \Pr[f(\mathbf{x}_{\text{per}} = c_B)]$$
(C.12)

We proceed by constructing the following two half-spaces:

$$A := \left\{ \boldsymbol{z} : \boldsymbol{\xi}^T C_{\sigma}^{-1} (\boldsymbol{z} - \boldsymbol{x}) \le \left[ \sqrt{\boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{\xi}} \right] \boldsymbol{\Phi}^{-1}(\rho_A) \right\}$$
(C.13)

$$B := \left\{ \boldsymbol{z} : \boldsymbol{\xi}^T C_{\sigma}^{-1} (\boldsymbol{z} - \boldsymbol{x}) \ge \left[ \sqrt{\boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{\xi}} \right] \boldsymbol{\Phi}^{-1} (1 - \rho_B) \right\}$$
(C.14)

Sec. C.2 shows that  $\Pr[\mathbf{x}_{nat} \in A] = \rho_A$ . Therefore, using Eq. (C.11) we know that  $\Pr[f(\mathbf{x}_{nat} = c_A)] \ge \Pr[\mathbf{x}_{nat} \in A]$ . Hence, we can apply Lemma C.2 with  $h(\mathbf{z}) := \mathbf{1}[f(\mathbf{z}) = c_A]$  to conclude:

$$\Pr[f(\mathbf{x}_{\text{per}} = c_A)] \ge \Pr[\mathbf{x}_{\text{per}} \in A]$$
(C.15)

Similarly, Sec. C.2 shows that  $\Pr[\mathbf{x}_{nat} \in B] = \rho_B$ . Therefore, using Eq. (C.11) we know that  $\Pr[f(\mathbf{x}_{nat} = c_B)] \leq \Pr[\mathbf{x}_{nat} \in B]$ . Hence, we can apply Lemma C.2 with  $h(\mathbf{z}) := \mathbf{1}[f(\mathbf{z}) = c_B]$  to conclude:

$$\Pr[f(\mathbf{x}_{\text{per}} = c_B)] \le \Pr[\mathbf{x}_{\text{per}} \in B]$$
(C.16)

Now, to prove Eq. (C.12), we need to show that  $\Pr[\mathbf{x}_{per} \in A] > \Pr[\mathbf{x}_{per} \in B]$ , since that would complete the following chain of inequalities:

$$\Pr[f(\mathbf{x}_{\text{per}} = c_A)] \ge \Pr[\mathbf{x}_{\text{per}} \in A] > \Pr[\mathbf{x}_{\text{per}} \in B] \ge \Pr[f(\mathbf{x}_{\text{per}} = c_B)] \quad (C.17)$$

We can compute the following:

$$\Pr[\mathbf{x}_{\text{per}} \in A] = \mathbf{\Phi}\left(\mathbf{\Phi}^{-1}(\rho_A) - \sqrt{\boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{\xi}}\right)$$
(C.18)

$$\Pr[\mathbf{x}_{\text{per}} \in B] = \mathbf{\Phi}\left(\mathbf{\Phi}^{-1}(\rho_B) + \sqrt{\boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{\xi}}\right)$$
(C.19)

Finally, algebra shows that  $\Pr[\mathbf{x}_{per} \in A] > \Pr[\mathbf{x}_{per} \in B]$  iff:

$$\sqrt{\boldsymbol{\xi}^T C_{\sigma}^{-1} \boldsymbol{\xi}} < \frac{\boldsymbol{\Phi}^{-1}(\rho_A) - \boldsymbol{\Phi}^{-1}(\rho_B)}{2}$$
(C.20)

which concludes the proof.

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## Deferred Algebra

First, we compute  $\Pr[\mathbf{x}_{nat} \in A]$  below:

$$\Pr[\mathbf{x}_{nat} \in A] = \Pr\left\{\boldsymbol{\xi}^{T} C_{\sigma}^{-1}(\mathbf{n}) \leq \left[\sqrt{\boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}}\right] \boldsymbol{\Phi}^{-1}(\rho_{A})\right\}$$
$$= \Pr\left\{\boldsymbol{\xi}^{T} C_{\sigma}^{-1} \mathcal{N}(0, C_{\sigma}) \leq \left[\sqrt{\boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}}\right] \boldsymbol{\Phi}^{-1}(\rho_{A})\right\}$$
$$= \Pr\left\{\boldsymbol{\xi}^{T} \mathcal{N}(0, C_{\sigma}^{-1} C_{\sigma}(C_{\sigma}^{-1})^{T}) \leq \left[\sqrt{\boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}}\right] \boldsymbol{\Phi}^{-1}(\rho_{A})\right\}$$
$$= \Pr\left\{\mathcal{N}(0, \boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}) \leq \left[\sqrt{\boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}}\right] \boldsymbol{\Phi}^{-1}(\rho_{A})\right\}$$
$$= \Pr\left\{\left[\sqrt{\boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}}\right] \mathcal{N}(0, 1) \leq \left[\sqrt{\boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}}\right] \boldsymbol{\Phi}^{-1}(\rho_{A})\right\}$$
$$= \Pr\left\{\mathcal{N}(0, 1) \leq \boldsymbol{\Phi}^{-1}(\rho_{A})\right\} = \boldsymbol{\Phi}\left(\boldsymbol{\Phi}^{-1}(\rho_{A})\right) = \rho_{A}$$

Similarly, now we compute  $\Pr[\mathbf{x}_{nat} \in B]$ :

$$\begin{aligned} \Pr[\mathbf{x}_{\text{nat}} \in B] &= \Pr\left\{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}(\mathbf{n}) \geq \left[\sqrt{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}\right] \boldsymbol{\Phi}^{-1}(1-\rho_{B})\right\} \\ &= \Pr\left\{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\mathcal{N}(0,C_{\sigma}) \geq \left[\sqrt{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}\right] \boldsymbol{\Phi}^{-1}(1-\rho_{B})\right\} \\ &= \Pr\left\{\boldsymbol{\xi}^{T}\mathcal{N}(0,C_{\sigma}^{-1}C_{\sigma}(C_{\sigma}^{-1})^{T}) \geq \left[\sqrt{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}\right] \boldsymbol{\Phi}^{-1}(1-\rho_{B})\right\} \\ &= \Pr\left\{\mathcal{N}(0,\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}) \geq \left[\sqrt{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}\right] \boldsymbol{\Phi}^{-1}(1-\rho_{B})\right\} \\ &= \Pr\left\{\left[\sqrt{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}\right]\mathcal{N}(0,1) \geq \left[\sqrt{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}\right] \boldsymbol{\Phi}^{-1}(1-\rho_{B})\right\} \\ &= \Pr\left\{\mathcal{N}(0,1) \geq \boldsymbol{\Phi}^{-1}(1-\rho_{B})\right\} = 1 - \boldsymbol{\Phi}\left(\boldsymbol{\Phi}^{-1}(1-\rho_{B})\right) = \rho_{B}\end{aligned}$$

Now, we compute  $\Pr[\mathbf{x}_{per} \in A]$ :

$$\Pr[\mathbf{x}_{nat} \in A] = \Pr\left\{\boldsymbol{\xi}^{T} C_{\sigma}^{-1}(\mathbf{n} + \boldsymbol{\xi}) \leq \left[\sqrt{\boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}}\right] \boldsymbol{\Phi}^{-1}(\rho_{A})\right\}$$
$$= \Pr\left\{\boldsymbol{\xi}^{T} C_{\sigma}^{-1} \mathcal{N}(0, C_{\sigma}) \leq \left[\sqrt{\boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}}\right] \boldsymbol{\Phi}^{-1}(\rho_{A}) - \boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}\right\}$$
$$= \Pr\left\{\mathcal{N}(0, \boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}) \leq \left[\sqrt{\boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}}\right] \boldsymbol{\Phi}^{-1}(\rho_{A}) - \boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}\right\}$$
$$= \Pr\left\{\left[\sqrt{\boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}}\right] \mathcal{N}(0, 1) \leq \left[\sqrt{\boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}}\right] \left(\boldsymbol{\Phi}^{-1}(\rho_{A}) - \sqrt{\boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}}\right)\right\}$$
$$= \Pr\left\{\mathcal{N}(0, 1) \leq \boldsymbol{\Phi}^{-1}(\rho_{A}) - \sqrt{\boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}}\right\} = \boldsymbol{\Phi}\left(\boldsymbol{\Phi}^{-1}(\rho_{A}) - \sqrt{\boldsymbol{\xi}^{T} C_{\sigma}^{-1} \boldsymbol{\xi}}\right)$$

Finally, we compute  $\Pr[\mathbf{x}_{per} \in B]$ :

$$\Pr[\mathbf{x}_{\text{per}} \in B] = \Pr\left\{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}(\mathbf{n} + \boldsymbol{\xi}) \ge \left[\sqrt{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}\right]\boldsymbol{\Phi}^{-1}(1 - \rho_{B})\right\}$$
$$= \Pr\left\{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\mathcal{N}(0, C_{\sigma}) \ge \left[\sqrt{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}\right]\boldsymbol{\Phi}^{-1}(1 - \rho_{B}) - \boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}\right\}$$
$$= \Pr\left\{\mathcal{N}(0, \boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}) \ge \left[\sqrt{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}\right]\boldsymbol{\Phi}^{-1}(1 - \rho_{B}) - \boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}\right\}$$
$$= \Pr\left\{\left[\sqrt{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}\right]\mathcal{N}(0, 1) \ge \left[\sqrt{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}\right]\boldsymbol{\Phi}^{-1}(1 - \rho_{B}) - \boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}\right\}$$
$$= \Pr\left\{\mathcal{N}(0, 1) \ge \boldsymbol{\Phi}^{-1}(1 - \rho_{B}) - \sqrt{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}\right\}$$
$$= \Pr\left\{\mathcal{N}(0, 1) \le \boldsymbol{\Phi}^{-1}(\rho_{B}) + \sqrt{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}\right\} = \boldsymbol{\Phi}\left(\boldsymbol{\Phi}^{-1}(\rho_{B}) + \sqrt{\boldsymbol{\xi}^{T}C_{\sigma}^{-1}\boldsymbol{\xi}}\right)$$

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