DEVELOPMENT AND EVALUATION OF A NOVEL NEURAL NETWORK OF PMSM FOR ELECTRIC VEHICLE

by

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A THESIS

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ABSTRACT

This thesis investigates an artificial neural network (ANN)-based field-oriented control (FOC) for a surface-mounted and an interior-mounted permanent magnet synchronous machine (SPMSM and IPMSM). The ANN was trained by using Levenberg-Marquardt and forward accumulation through time algorithm.

First, the thesis examines the fundamentals of motor parameters and two aforementioned vector controls, with training algorithms, in detail. Then, the background and various algorithms of Maximum Torque per Ampere (MTPA) and flux weakening (FW) control are undertaken while the following part epitomizes an off-the-shelf component-based electric vehicle (EV) model that is constructed using MATLAB SimPowerSystems and SimDriveline.

The proposed control is validated in both simulation and hardware experiment and compared with a PI-based field-oriented control. First, for SPMSM, the results of simulation and hardware experiment show that the maximum operating speed of the proposed control is improved by 48% and 3.5% compared to the PI-based control. For IPMSM, the results show that the proposed control produces less d-axis current than the latter control.

Moreover, the control is implemented and simulated in electric vehicle model, which is constructed using SimPowerSystems and SimDriveline library in Simulink by the author with off-the-shelf components. The results show that the proposed controller can be a potential replacement of the existing control schemes, such as PID, fuzzy logic, or others, and provides adequate traction control in EV application.
DEDICATION

This thesis is dedicated to everyone who helped me and guided me through the process of creating this document. Among many contributors, such as my family, Dr. Li, and Xingang Fu, I like to truly give this glory to J. C.
LIST OF ABBREVIATIONS AND SYMBOLS

PMSM  Permanent Magnet Synchronous Machine

$mfm$  Magneto-motive Force

EMF  Electromagnetic-motive Force

$\vec{F}_a(t)$  Space vector magneto-motive force in the air gap, where “a” denotes the reference stator a-axis with an angle of 0° while “→” illustrates an instantaneous quantity

$e^{j\theta}$  $1 \angle \theta = \cos \theta + j \sin \theta$

$\theta_{da}$  Angle between d- and a-axis

$\vec{i}_d(t)$  Space vector stator current with the d-axis as the reference axis

$i_{a,b,c}$  Phase a, b, and c current

$[T_s]_{abc \rightarrow dq}$  Matrix that changes abc to dq reference frame (Park’s transformation)

$[T_s]_{dq \rightarrow abc}$  Matrix that changes dq to abc reference frame (inverse Park’s transformation)

$R_{coi}$  Coil resistance

$T_c$  Number of turns in the coil

$\rho$  Electrical resistivity [Ω·m]

$L$  Mean turn length [m]

$n$  Number of parallel strands in each conductor

$A$  Cross-sectional area of one strand [m²]

$N_c$  Number of multiple coils per phase
\( R_{\text{ph}} \) Phase resistance
\( \alpha_{\text{parallel}} \) Parallel paths
\( \rho_0 \) Electrical resistivity at 20°C
\( T \) Temperature
\( T_0 \) 20°C
\( e \) Induced voltage
\( L_{g, \text{slot, end}} \) Air-gap, stator slot-leakage, and end-turn inductance
\( M_{g, \sigma} \) Air-gap and leakage mutual inductance
\( R_s \) Stator phase resistance
\( L_{-\text{sd,sq}} \) Stator d- and q-axis inductance
\( \lambda_{\text{sd,sq}} \) Stator d- and q-axis flux linkage
\( v_{\text{sd,sq}} \) Stator d- and q-axis voltage
\( i_{\text{sd,sq}} \) Stator d- and q-axis current
\( K_t \) Torque constant
\( \lambda_{f\text{d}} \) Flux linkage constant
\( K_e \) Back-emf constant
\( p \) Number of pole pairs
\( \omega_m \) Mechanical angular velocity
\( \omega_r \) or \( \omega_c \) Electrical angular velocity (\( \omega_r = p * \omega_m \))
\( J_{\text{eq}} \) Motor inertia
\( T_{\text{em}} \) Electromagnetic torque
\( T_{\text{load}} \) Load torque
\( U(\cdot) \) Utility function
\( C(\cdot) \) \hspace{1em} \text{DP Cost function}

\( e_{d,q} \) \hspace{1em} \text{Error function of dq current}

\( s_{d,q} \) \hspace{1em} \text{Integrals of the error terms}

\( \gamma \) \hspace{1em} \text{Discount factor}

\( J \) \hspace{1em} \text{Jacobian matrix}

\( p \) \hspace{1em} \text{Parameter vector}

\( x \) \hspace{1em} \text{Estimated measurement vector}

\( \delta_p \) \hspace{1em} \text{Minimization function}

\( \frac{\partial C}{\partial w} \) \hspace{1em} \text{Gradient of weight vector}

\( V(\cdot) \) \hspace{1em} \text{Error function for pattern}

\( \mu \) \hspace{1em} \text{Parameter that determines the steepest descent algorithm and Gauss Newton}

\( \text{Epoch}_{\text{max}} \) \hspace{1em} \text{Maximum of Epoch}

\( \beta_{\text{de, in}} \) \hspace{1em} \text{Decreasing and increasing factor}

\( \mu_{\text{max}} \) \hspace{1em} \text{Maximum acceptable } \mu

\( \left\| \frac{\partial C}{\partial w} \right\|_{\text{min}} \) \hspace{1em} \text{Norm of the minimum acceptable gradient}

\( \text{LM} \) \hspace{1em} \text{Levenberg-Marquardt}

\( \text{FATT} \) \hspace{1em} \text{Forward accumulation through time}

\( \text{FOC} \) \hspace{1em} \text{Field-oriented Control}

\( \text{PI} \) \hspace{1em} \text{Proportional-Integral}

\( \text{PM} \) \hspace{1em} \text{Permanent Magnet}

\( \text{RNN} \) \hspace{1em} \text{Recurrent Neural Network}

\( \text{ANN} \) \hspace{1em} \text{Artificial Neural Network}

\( \text{NN} \) \hspace{1em} \text{Neural Network}
<table>
<thead>
<tr>
<th>Abbreviation</th>
<th>Full Form</th>
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<tbody>
<tr>
<td>DP</td>
<td>Dynamic Programming</td>
</tr>
<tr>
<td>IPMSM</td>
<td>Interior mounted PMSM</td>
</tr>
<tr>
<td>SPMSM</td>
<td>Surface mounted PMSM</td>
</tr>
<tr>
<td>PWM</td>
<td>Pulse Width Modulation</td>
</tr>
<tr>
<td>SPWM</td>
<td>Sinusoidal PWM</td>
</tr>
<tr>
<td>SVPWM</td>
<td>Space Vector PWM</td>
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</table>
ACKNOWLEDGMENTS

If I was only one to prepare this thesis, I would like to inform that this thesis will not be done. I like to express my gratitude to my family, colleagues, friends, mentors, and faculty advisors. Further, the one who gave this topic to research and support was Dr. Shuhui Li, associate professor and the chairman of this thesis. Without his support and guide, I couldn’t learn and have the experiences that were provided throughout this thesis. In addition, I would like to appreciate the support and care from Dr. Xingang Fu, who trained and gave guidelines to the neural network control. Lastly, I would like to express my appreciation to all faculty members. They provided the academic material and led me to pursue the Masters.

My family waited two years, even though it would have been one year, to complete this thesis. When I was depressed that I couldn’t finish this thesis by one year, they encouraged me by saying “keep it up”.

Among the contributors and supporters, the most support came from J.C.
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CHAPTER 1
INTRODUCTION

Among various AC machines, permanent magnet synchronous motors (PMSMs) gained the most attention due to the following advantages: high torque-inertia ratio, high efficiency, high power density, and compact structure [1, 2]. These advantages led PMSMs to be utilized in various applications such as air conditioner, refrigerator, direct-drive washing machines, automotive electrical power steering, machine tools, traction control, and data storage applications [3]. Furthermore, with the advent of the hybrid electric vehicle and electric vehicle, the demand for PMSMs has grown exponentially. Other machines, such as induction motor and synchronous reluctance motor, also received attention, but due to its high torque density, less weight, high dynamic performance under load, high efficiency, and the absence of heat, PMSMs dominated the market [4]. As Table 1.1 suggests, four out of five vehicles are using PMSM in its application.

<table>
<thead>
<tr>
<th>Vehicle</th>
<th>Type of motor</th>
</tr>
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<tbody>
<tr>
<td>Toyota Prius</td>
<td>Permanent magnet AC synchronous motor [5]</td>
</tr>
<tr>
<td>Tesla Model S</td>
<td>Four pole AC induction motor [6]</td>
</tr>
<tr>
<td>Hyundai Sonata hybrid</td>
<td>Interior permanent magnet synchronous motor [7]</td>
</tr>
<tr>
<td>Nissan Leaf EV</td>
<td>AC synchronous electric motor [8]</td>
</tr>
<tr>
<td>Honda Accord hybrid</td>
<td>AC synchronous permanent-magnet electric motor [9]</td>
</tr>
</tbody>
</table>

Table 1.1: Types of motor in various vehicles

In DC motor, the stator and rotor can be excited independently. Usually, the value of flux can be controlled by applying the current through the stator while the torque can be controlled by manipulating the current through the rotor windings. On the other hand, in AC motor, the stator
and rotor cannot be excited independently. Generally, only source that can be controlled is the stator currents. However, with the advent of Field Oriented Control (FOC), the decoupled control of torque and flux in AC machines became possible by transforming alternating three-phase currents into DC two-phase currents. This algorithm not only produces high efficiency in a wide range of speeds, but also makes motor control easier in both low speed and high speed.

1.1 Literature Review

In general, the current industries use a PI-based field-oriented control to operate PMSMs. The advantage of the PI controller is its simplicity and ease of understanding. However, recent studies show that such control strategy is experiencing the following bottlenecks: a) difficulty to obtain the optimal PI coefficients that satisfy both stability and performance, b) no true decoupling between torque and flux, and c) limited stability and performance [1, 10]. To overcome these obstacles, various control schemes were proposed.

Tzann-Shin Lee et al. [1] proposed an adaptive $H_{\infty}$ controller to enhance the stability in system under parameter perturbations and external disturbances. To accomplish the task, a quadratic storage function was chosen with a state feedback term to produce a simple adaptive law for the controller. The study applied periodic step and sinusoidal commands to test its controller. Although the study proved its effectiveness under the preceding conditions, it did not show any result when the motor was in high speed or random speeds.

Kamel et al. [11] utilized a fuzzy logic controller with 49 rules to drive the PMSM and compared the results with the conventional PI-based vector controller. The result showed that the proposed fuzzy logic controller tracked the speed reference faster and stabilized faster when load torque disturbance occurred. However, the evaluation was verified only in simulation.
Lu et al. [12] employed a self-constructing recurrent fuzzy neural network (SCRFNN) to control the PMSM online. The proposed SCRFNN was incorporated with two control schemes: self-constructing fuzzy neural network and recurrent neural network. Even though the proposed control showed good performance, the validation was only done in simulation. In addition, the simplified mathematical model of PMSM was used for simulation.

Xia et al. [13] presented a novel direct torque and flux control for matrix converter based PMSM drive. The study investigated all matrix converter switching states to control input and output variable. Four switching schemes were based on the tables, which have optimal selections of switching. Compared with the conventional PI based controller, the matrix converter based PMSM drive achieved to reduce 30% of the standard deviations of torque and flux and 9% of the total harmonic distortion. However, even though the result shows improvement, implementing this technology in industry is difficult because most industries utilize either IGBT or MOSFET based inverter fed drive.

Khan et al [14] proposed a novel neuro-network-based self-tuned controller for IPMSM drives. With having the speed error and its change as inputs, the weights, translation, and dilation of the wavelet functions were altered. Simultaneously, a back propagation training algorithm continuously manipulated the controller parameter. Additionally, the Lyapunov stability was used to enhance the system robustness and stability. The controller was validated through both simulation and hardware and compared with PI-based controller. The author asserted that the proposed controller produced zero speed overshoot and undershoot and performed better against load changes than PI-controller. However, only minimal improvement was observed.

J. Moreno et al [15] applied a neural network control in hybrid vehicle energy management system to minimize the discharge of the batteries. A test vehicle was equipped with
two primary energy sources: lead-acid batteries and ultracapacitor bank. The result showed that the NN control achieved 8.9% increase in range. The experiment was validated in both simulation and hardware.

Z. Chen et al [16] utilized two NN modules that are trained by dynamic programming to optimally manage the energy between engine and battery in power-split hybrid electric vehicle. This control was effective when the trip length and trip duration were known before the trip starts. The proposed technology was validated only in simulation.

Figure 1.1 shows the chart of the current publications relating to PMSM controls, where the values in parentheses are referring to the total publications, including conference and journal papers. As it illustrates, only few papers covered PMSMs in wide speed range and automotive applications.

![Figure 1.1: Chart of current publications](image)

Figure 1.1: Chart of current publications
1.2 Research Motivation

The literature search suggests that there is no research conducted on an artificial neural network-based vector control for PMSM including both flux weakening control and EV implementation. Thus, this thesis proposes the neural network vector control for both SPMSM and IPMSM with flux weakening control. The SPMSM motor drive with the proposed control will be validated through both simulation and experiment. Further, the IPMSM will be implemented in EV model that the author has made using the MATLAB tools.

1.3 Thesis Organization

The thesis is organized as follows. Chapter 2 gives the full descriptions of PMSM by explaining the motor parameters that affect the control while Chapter 3 introduces the conventional PI-based vector control and proposed neural network vector control with full descriptions on training algorithm. The next chapter discusses a maximum torque per ampere (MTPA) control algorithm and flux weakening control, whereas Chapter 5 covers the EV simulation model that was constructed using SimPowerSystems and SimDriveline. Chapter 6 shows the simulation result of the NN control for SPMSM, IPMSM, and IPMSM in EV application while Chapter 7 presents the hardware result of SPMSM. Finally, the conclusion of the thesis is provided in Chapter 8.
CHAPTER 2
MODELLING PMSM

Before investigating the controls for PMSMs, this section briefly summarizes the motor parameters that affect the control. First, a background of PM machines is epitomized, and the second subsection discusses the fundamentals of the motor parameter, such as stator resistance, inductances, flux linkages, associated constants, motor inertia, and number of poles. In the final subsection, the mathematical model and Simulink model of PMSM is covered.

2.1 Background Information of PM motor

Many electrical machine literatures had classified and organized PM machines into different families to improve understanding of similarities and differences, associated with PM machines. In this text, the author has chosen to organize the machines according to 1) the direction of field distribution, 2) configuration of stator winding, and 3) physical structure of rotor.

In general, depending on the direction of the field distribution, PM machines are divided into two types. If radial-directed fields are in the airgap, it is called radial-flux PM machine while if the field in the airgap is axial direction, it is called axial-flux or “pancake” PM machine. In current EV application, the radial-flux machines are utilized more than the axial-flux machine because the former machine has easier manufacturing process and more space flexibility than the latter. Thereby, this thesis chose to investigate the radial-flux machine.
Within the radial-flux PM machine, there are two types of stator winding: distributed and concentrated. A distributed wounded stator winding produces a sinusoidally distributed field in the air gap while a concentrated stator winding produces a trapezoidal distributed field distribution. If the stator winding is distributed winding, it is called PMSM or PMAC while if the stator winding is concentrated winding, it is called brushless DC motor (BLDC). Out of two, the most commonly used radial-flux machine is PMSM because it is much easier to produce the sinusoidal waveform than the trapezoidal waveform. Furthermore, FOC can be implemented only for PMSM because FOC cannot interpret the trapezoidal fields. Thus, in this thesis, the PM machine that is equipped with distributed stator winding was used.

Lastly, the radial PM machines are divided further by the location of magnets in the rotor. If magnets are on the surface of the rotor, it is called surface-mounted PM machines (SPMSMs), whereas interior-mounted PM machines (IPMSMs) has magnets inside the rotor. This context has investigated both machines because both machines are widely used for various applications.

2.2 PM Motor Parameter

For overall simulation, the PMSM model from MATLAB/ SimPowerSystems is used. Figure 2.1a shows the high-level PMSM model in MATLAB/SimPowerSystems while Fig. 2.1b represents the parameters that are necessitated by the model. The main reason for employing this block is that it not only provides the interface between other components in SimPowerSystems, but also has easy GUI to be used. As Fig. 2.1b illustrates, the parameters that are required by the model are 1) stator resistance, 2) armature inductance, 3) flux linkage constant, 4) inertia, and 5) motor pole pairs.
Figure 2.1: High level a) model of, b) parameters of the PMSM in SimPowerSystems
2.2.1 Resistance

The primary purpose of the stator is producing the rotating magnetic field in the airgap to rotate the rotor. In the stator, the rotating magnetic field can be produced by applying the current through the stator coil. For every coil, there is an intrinsic resistivity associated with it. Table 2.1 shows the intrinsic resistivity of enameled copper, aluminum, and silver [17].

<table>
<thead>
<tr>
<th>Material</th>
<th>Resistivity (Ω·m) at 20 ºC</th>
</tr>
</thead>
<tbody>
<tr>
<td>Enameled copper</td>
<td>$1.68 \times 10^{-8}$</td>
</tr>
<tr>
<td>Aluminum</td>
<td>$2.82 \times 10^{-8}$</td>
</tr>
<tr>
<td>Silver</td>
<td>$1.59 \times 10^{-8}$</td>
</tr>
</tbody>
</table>

Table 2.1: Electrical resistivity of copper, aluminum, and silver

By using the resistivity, the resistance of the coil can be calculated by

$$R_{\text{coil}} = \rho \frac{l}{A},$$

(2-1)

where $R_{\text{coil}}$ is the electrical resistance of single coil [Ω]; $\rho$ is the electrical resistivity [Ω·m]; $l$ is the length of the coil [m]; and $A$ is the cross-sectional area of the coil [m²]. Since the stator will have multiple turns per phase, the phase resistance is

$$R_{\text{ph}} = N_c \cdot R_{\text{coil}},$$

(2-2)

where $N_c$ is the number of turns.

There are two ways to measure the resistance using either modern multimeter or LCR meter: two-wire technique and four-terminal resistance measurement. Generally, the latter method is more accurate than the former method because the latter method excludes the resistance of the test lead from the actual measurement. Thus, this thesis will elucidate the instructions of four-terminal resistance measurement.

Usually, the stator winding is connected in an internal “wye” connection, which causes the neutral point to be hidden in the stator, and thus inaccessible. Thereby, to measure the stator resistance, two phase resistances have to be measured. Figure 2.2 shows the aforementioned
measuring connection while Figure 2.3 illustrates the schematic of four-terminal measurement. As Fig. 2.2 shows, a positive probe is connected with A-phase while a negative probe with C-phase. Note that putting the probe is on other phases are allowed.

Figure 2.2: Measuring the stator resistance using digital multimeter

Figure 2.3: Four-terminal measurement schematic
2.2.2 Inductance

The synchronous inductance of PMSMs is denoting the inductance of a winding in the synchronous reference frame. If the reference frame is changed from a stationary reference frame to a rotating reference frame, the synchronous inductance is divided into two inductances: d- and q-axis inductance. The d-and q-axis inductance can be found by using these equations.

\[
L_d = \frac{2}{3}L, (rotor \ angle = 0^\circ) \tag{2-3}
\]

\[
L_q = \frac{2}{3}L, (rotor \ angle = 90^\circ) \tag{2-4}
\]

in which \(L\) is the synchronous inductance while \(L_d\) and \(L_q\) denotes the d- and q-axis inductance [18]. More information about transforming reference frames can be found in Section 2.3.

Generally, the location of magnets determines the d- and q-axis inductances. If PM motor has its magnets on the surface of the rotor (SPMSM), the inductances of d and q-axis are same while if the PM motor has its PMs in the rotor (IPMSM), the inductances are different for d and q-axis. Figure 2.4 represents the configuration of surface and interior mounted PM motors.

![Figure 2.4: a) Surface mounted magnets, b) Inset rotor magnets, c) Buried tangential magnets, d) Spoke-type magnets, e) V-shape magnets, and f) Multilayer V-shape magnets PMSM [19]](image-url)
The inductances affect the performance of the motor as follows [20].

1. The inductive voltage drop absorbs a fraction of the supply. Thus, the base and maximum speed drops.

2. In sinewave, the inductance affects the power factor. The bigger the inductance is, the more reactive power has to be produced.

3. It distorts the excitation flux of the magnets, changes the torque constant nonlinearly, and causes the armature reaction when the stator current is flowing through the conductor.

4. It also hinders the power electronic drives.

5. Lastly, there is possibility of short-circuit fault error.

To measure the d- and q- axis inductances, four equipment are required. They are 1) DC power supply, 2) oscilloscope, 3) current probes, and 4) voltage probe. First, to measure the d-axis inductance, the rotor needs to be aligned to Phase A. One can accomplish that by connecting the positive potential of power supply to Phase A terminal and the negative potential to Phase B or C terminal. Then, turn on the power supply and increase the voltage slightly. After alignment, lock the rotor. Then, one needs to connect Phase B and C terminals to the positive potential while ground Phase A. This connection allows to apply negative voltage across the circuit. Then, apply a current probe on conductor that connects Phase A and positive potential. After checking previous connections, turn on the power supply [21].

In real environment, if the step voltage is applied, it will look like blue line and purple line in Fig. 2.6a and 2.6b, respectively. The current will behave like green curve in Fig. 2.6b. Then, one can determine the d-axis inductance by using this equation [21].
\[ l_d = \frac{V}{R} \left( 1 - e^{-\frac{V}{R}} \right), \tau = \frac{L_d}{R} \] (2-5)

where \( V \) is the amplitude of the applied voltage and \( R \) is the line-to-line resistance. Likewise, the q-axis inductance can be measured, except the rotor needs to be aligned to q-axis. This can be done by connecting Phase B to the positive terminal while Phase C to the negative terminal. Phase A is not used.

Figure 2.5: Inductance measurement schematic circuit
2.2.3 Back-EMF Constant, Flux Linkage Constant, and Torque Constant

Back-EMF, torque, and flux linkage constants are important constants for evaluating machine performance. First, the back-EMF constant describes the behavior of back-EMF with respect to mechanical speed of the machine. Generally, the units of the back-EMF constant are in Volts line-to-line (V_{L-L})/Revolution per Minutes (RPM). Some manufacturers use Volts root mean square (V_{RMS})/kRPM. Second, the torque constant describes the output torque per current. The units of the torque constant are in Nm/A_{RMS}. Again, some manufacturers use Nm/A_{peak}. Third, the flux linkage constant describes the flux linkage that is produced by permanent magnet on or in the rotor. The units of this constant are in Volts·Seconds (V·s). At first glance, these
three terms are not interrelated to each other, but they are. These constants can be equated to each other using proper gains like shown in below equations.

\[
K_e = \frac{V}{\text{rpm}} = \frac{30}{\pi} \cdot \frac{V}{\text{rad/sec}} = \left(\frac{kgm^2}{A_{\text{rms}sec^2}}\right) = V \cdot \text{sec}
\]

(2-6)

\[
K_t = \frac{Nm}{A_{\text{rms}}} = \left(\frac{kgm^2}{A_{\text{rms}sec^2}}\right) = V \cdot \text{sec}
\]

(2-7)

\[
\lambda_{fa} = \frac{V}{\text{rad/sec}} = V \cdot \text{sec}
\]

(2-8)

As illustrated, since all motor constants are correlated to each other, knowing one motor constant can calculate others without difficulty. Thus, this thesis will mainly cover the measurement method for back-EMF. However, to give more measuring options to readers, this thesis will also include the measuring method for torque constant.

First, to measure the back-EMF constant of the machine, an extra motor and an oscilloscope are needed. Since the primary objective of the extra motor is running the test motor with constant speed, any kinds of motor can be used. The first step is connecting the auxiliary and test motor through the mechanical shaft as shown in Fig. 7.8. The next step is rotating the test motor with certain constant speed. Then, connect positive and negative probe to any phase terminal while other end of the probe with the oscilloscope. This will measure the peak line-to-line voltage. The back-EMF constant can be calculated by dividing the measured voltage with the speed [18].

Second, to measure the torque constant, the only required resources are the motor torque vs. speed graph and motor current vs. speed graph. Usually, the manufacturer provides these data. The torque constant of the machine can be calculated by dividing the torque with the current at specific speed. If the units of the current are in RMS, a peak value can be calculated by multiplying \(\sqrt{2}\) to the RMS value. Figure 2.7a shows the typical torque and speed curve, whereas
2.2.4 Mechanical Parameters and Number of Poles

The concept of mechanical parameters of machines begins with the electrodynamic equation.

\[ J_{eq} \frac{d\omega_r}{dt} = T_{em} - b\omega_r - J_o - T_{load} \]  (2-9)

in which \( J_{eq} \) is the inertia of the machine; \( \omega_r \) is the mechanical speed of the motor; \( T_{em} \) is the electromagnetic torque from the motor; \( b \) is the viscous damping coefficient; \( J_o \) is the coulomb friction constant; and \( T_{load} \) is the external load torque. In this equation, there are three mechanical parameters: inertia, viscous damping coefficient, and coulomb friction constant.

First, the inertia of PM motors determines the performance between the mechanical stability and acceleration and deceleration. Larger the inertia, better the mechanical stability is, but lesser acceleration and deceleration performance is. Further, the inertia is proportional to the size of the machine. Bigger the size, larger the inertia is. Generally, the units of the inertia are in kg·m².
Second, the viscous damping coefficient describes the energy dissipation due to the movement of the machine. Usually, it is related with vibration of the machine. The units of this coefficient are in Nm·second.

Lastly, the coulomb friction constant denotes the shaft static friction. In most cases, the coulomb friction constant is neglected in calculation and simulation because the impact of the shaft static friction is minimal. The units are in Nm.

In general, most manufacturers provide these mechanical parameters. However, using provided values for simulation and experiment may produce inaccurate results because these parameters are measured when the motor is not connected with other system. Thus, measuring the mechanical parameters with the complete system is needed for accurate result.

First, to calculate the inertia, determining \( b \) and \( J_0 \) is needed. These values can be measured by using the method in [21]. After identifying \( b \) and \( J_0 \), the inertia can be found by using a method called Coast down test. The test is really simple.

The first step is rotating the test motor to its maximum operating speed. Then, the next step is disconnecting the power by either unplug the cable or turn off the controller. The rotor speed will behave as in Figure 2.8. The final step is applying this equation to calculate the inertia.

\[
\omega_r = \left( \omega_{r0} + \frac{J_0}{b} \right) e^{-\frac{b}{Ie}} - \frac{J_0}{b} \quad (2-10)
\]

As it shows, at \( t = 0.2 \) sec, the power is off. Then, the motor is spun down to a stop at 0.6 sec.
The number of poles defines a ratio between electrical and mechanical quantities (mechanical vs. electrical rotor position/angular speed). Thus, determining the number of poles in machines is important. Usually, the number of poles is written on the label of the motor. However, if the manufacturer does not provide this information, one can determine the number of poles by using back-EMF test that was discussed earlier in section 2.2.3. Figure 2.9 shows the typical back-EMF vs. time graph. The number of poles can be calculated using this equation.

\[
P = \frac{2\omega_e}{\omega_m} = \frac{2T_r}{T_e}
\]  

(2-11)
2.3 Park’s Transformation and Inverse Park’s Transformation

Park’s transformation triggered the AC machine to gain attentions from the world because it allowed to control AC machine like controlling DC machine by reducing the three AC quantities to two DC quantities.

The concept of Park’s transformation begins with Clarke’s transformation, which converts abc-winding into $\alpha\beta$-quadrature winding.

$$\begin{bmatrix} v_\alpha(t) \\ v_\beta(t) \end{bmatrix} = \begin{bmatrix} 1 & \cos(\gamma) & \cos(2\gamma) \\ 0 & \sin(\gamma) & \sin(2\gamma) \end{bmatrix} \begin{bmatrix} i_\alpha(t) \\ i_\beta(t) \\ i_\gamma(t) \end{bmatrix}, \text{ where } \gamma = \frac{2\pi}{3}$$  \hspace{1cm} (2-12)
As the equation exhibits, the outputs are still time dependent. This can be solved by using Park’s transformation, which transforms αβ-quadrature winding stationary reference frame into rotating reference frame.

\[
\begin{bmatrix}
    i_d \\
    i_q
\end{bmatrix} = \begin{bmatrix}
    \cos(\theta_{da}) & \sin(\theta_{da}) \\
    -\sin(\theta_{da}) & \cos(\theta_{da})
\end{bmatrix} \begin{bmatrix}
    i_\alpha(t) \\
    i_\beta(t)
\end{bmatrix}, \text{where } \theta = \omega t
\] (2-13)

where \( \theta_{da} \) is the angle between the a-axis and d-axis. If Eq. 2-12 and 2-13 are combined, the resultant equation becomes

\[
\begin{bmatrix}
    i_d \\
    i_q
\end{bmatrix} = \sqrt{\frac{2}{3}} \begin{bmatrix}
    \cos(\theta_{da}) & \cos\left(\theta_{da} - \frac{2\pi}{3}\right) & \cos\left(\theta_{da} - \frac{4\pi}{3}\right) \\
    -\sin(\theta_{da}) & -\sin\left(\theta_{da} - \frac{2\pi}{3}\right) & -\sin\left(\theta_{da} - \frac{4\pi}{3}\right)
\end{bmatrix} \begin{bmatrix}
    i_\alpha(t) \\
    i_\beta(t) \\
    i_\gamma(t)
\end{bmatrix}
\] (2-14)

where Figure 2.10 illustrates Clarke’s transformation and Park’s transformation graphically.

Lastly, the equation that transforms dq axis to abc winding is called inverse Park’s transformation.

\[
\begin{bmatrix}
    i_a(t) \\
    i_b(t) \\
    i_c(t)
\end{bmatrix} = \sqrt{\frac{2}{3}} \begin{bmatrix}
    \cos(\theta_{da}) & \sin(\theta_{da}) \\
    \cos\left(\theta_{da} + \frac{4\pi}{3}\right) & -\sin\left(\theta_{da} + \frac{4\pi}{3}\right) \\
    \cos\left(\theta_{da} + \frac{2\pi}{3}\right) & -\sin\left(\theta_{da} + \frac{2\pi}{3}\right)
\end{bmatrix} \begin{bmatrix}
    i_{sd} \\
    i_{sq}
\end{bmatrix}
\] (2-15)

Figure 2.10: Reference frame [22]
2.4 Mathematical PMSM Model

The concept of the mathematical PMSM model begins with the stator d- and q-axis flux linkage equations, which are written as:

\[
\begin{bmatrix}
\lambda_{sd} \\
\lambda_{sq}
\end{bmatrix} = \begin{bmatrix}
L_{sd}i_{sd} + \lambda_{fd} \\
L_{sq}i_{sq}
\end{bmatrix}
\]  
(2-16)

where \( L_{sd, sq} \) are the stator d- and q-axis inductances and equal to \( L_{md, mq} + L_\ell \) while \( \lambda_{fd} \) is the flux linkage constant of the stator d-axis winding due to flux produced by the magnets in the rotor, where it is assumed that the d-axis is always aligned with the rotor magnetic axis, which is a-axis.

In terms of the aforementioned flux linkage equation, the stator winding voltages can be written as follows.

\[
\begin{bmatrix}
v_{sd} \\
v_{sq}
\end{bmatrix} = \begin{bmatrix}
R_s i_{sd}(t) + \frac{d}{dt} \lambda_{sd} - \omega_e \lambda_{sq} \\
R_s i_{sq}(t) + \frac{d}{dt} \lambda_{sq} + \omega_e \lambda_{sd}
\end{bmatrix}
\]  
(2-17)

in which \( \omega_e \) is the electrical angular velocity in electrical rad/s. The electrical angular velocity can be calculated by multiplying the electromechanical angular velocity, \( \omega_{\text{mech}} \), with the pole pairs, \( p \). If Eq. 2-16 and Eq. 2-17 are combined, the equation becomes

\[
\begin{bmatrix}
v_{sd} \\
v_{sq}
\end{bmatrix} = \begin{bmatrix}
R_s + L_{sd} P - \omega_e L_{sq} \\
\omega_e L_{sd} (R_s + L_{sq} P)
\end{bmatrix} \begin{bmatrix}
i_{sd} \\
i_{sq}
\end{bmatrix} + \begin{bmatrix}
0 \\
\lambda_{fd} \omega_e
\end{bmatrix}
\]  
(2-18)

where \( P \) is the derivative with respect to time. Under a balanced sinusoidal steady state condition, the terms associated with the time derivative are neglected. Thus, the dq currents become DC. As a result, Eq. 2-18 in steady state can be rewritten as

\[
\begin{bmatrix}
v_{sd} \\
v_{sq}
\end{bmatrix} = \begin{bmatrix}
R_s & - \omega_e L_q \\
\omega_e L_d & R_s
\end{bmatrix} \begin{bmatrix}
i_{sd} \\
i_{sq}
\end{bmatrix} + \begin{bmatrix}
0 \\
\lambda_{fd} \omega_e
\end{bmatrix}
\]  
(2-19)

The electromagnetic torque for asynchronous and synchronous machines is same. The equation is
\[ T_{em} = \frac{p}{2} (\lambda_{sd} i_{sq} - \lambda_{sq} i_{sd}) \]  

(2-20)

If Eq. 2-16 is substituted, the equation becomes

\[ T_{em} = \frac{p}{2} (\lambda_{fd} i_{sq} + (L_{sd} - L_{sq}) i_{sd} i_{sq}) \]  

(2-21)

2.5 Simulink PMSM model

Throughout the thesis, the PMSM model in SimPowerSystems was used because it not only reflects the mathematical representation that was derived in previous section, but also allows connecting with other electrical SimPowerSystems components, such as inverter and others. Thus, this section will summarize how the mathematical model that is explained in previous section is translated into block diagram.

As Fig. 2.11 shows, total of four blocks are used to represent the PM machine model:

1. Powersysdomain – Convert ABC SimPowerSystem voltage and current inputs to Simulink signal.
2. Electrical model – Possess the mathematical model in section 2.4.
3. Mechanical model – Calculate the mechanical speed and rotor angle using electrodynamics.
4. Measurement – Combine the associated parameters like speed, rotor angle, abc currents, and etc into one bus for simplicity.

Since the purpose of Powersysdomain and Measurement block are only translating SimPowerSystem signal to Simulink signal and organizing the calculated signal from other blocks, respectively, the detailed descriptions will not be covered.
First, ‘Electrical model’ subsystem is comprised of four blocks as shown in Figure 2.12. As illustrated, first, ‘abc2qd’ block transforms the three-phase voltage to the dq-winding voltages. Then, the dq-winding voltages are sent to a block called ‘iq, id’. The mask view of the ‘iq, id’ block is shown in Fig. 2.13 while Fig. 2.14 and 2.15 illustrate the internal components of the ‘iq’ and ‘id’ block, respectively. The equations in the ‘iq’ and ‘id’ block are derived from Eq. 2-18. However, in lieu of calculating the voltage, it is used to calculate the d and q-axis currents. The equations are

\[
\frac{d}{dt} i_{sq} = \frac{v_{sq} - R_s i_{sq} - \omega_q L_d i_{sd} - \omega_d A_f d}{L_q} \quad (2-22)
\]

\[
\frac{d}{dt} i_{sd} = \frac{v_{sd} - R_s i_{sd} + \omega_d L_q i_{sq}}{L_d} \quad (2-23)
\]

Then, the calculated d- and q-axis currents are applied to ‘qd2abc’ and ‘Te’ block for transforming the corresponding currents into three-phase and calculating the torque, respectively.
Figure 2.12: Block diagram of electrical model

Figure 2.13: Block diagram of iq,id subsystem
Second, the main purpose of ‘Mechanical model’ block is calculating the mechanical angular displacement and angular velocity. Figure 2.16 shows the mask view of ‘Mechanical model’ block. To calculate the angular velocity, the following equation is used.

\[
\frac{d}{dt} \omega_{mech} = \frac{T_{em} - T_{\text{load}}}{J_{eq}}
\]

(2-24)

where \( T_{em} \) is the electromagnetic torque that are calculated in electrical block; \( T_{\text{load}} \) is the load torque that is externally inserted; and \( J_{eq} \) is the motor inertia. The angular displacement of a rotor, \( \theta_{mech} \), can be calculated by integrating the angular velocity.

\[
\theta_{mech} = \int \omega_{mech}
\]

(2-25)
Figure 2.16: Block diagram of mechanical model
CHAPTER 3
CONVENTIONAL AND NN CONTROL PMSM

The previous section discussed the PM motor in detail from the motor parameters to Simulink model. This section introduces the control of PMSMs using a standard PI-based dq field-oriented vector control (FOC) and a recurrent neural network (RNN) control, which is trained by Levenberg-Marquardt (LM) algorithm and a forward accumulation through time (FATT) algorithm.

3.1 General Overview of FOC

The main objective, and also advantage, of FOC is separating the magnetic field and torque control. This independent control allows controlling AC machine as separately excited DC motor by aligning d-axis with rotor field flux linkage. Figure 3.1 illustrates how the vector control works in closed loop.

Figure 3.1: Overall block diagram of FOC for IPMs

FOC control has two distinctive nested-loop controllers: an outer speed-loop controller and an inner current-loop controller. The main role of the outer speed loop controller is ensuring
the actual speed of the motor to track the reference speed, whereas the inner current controller
does the same, except it regulates the current, not the speed. Further, to reject disturbances before
the signals propagate to the inner loop, the outer loop controller has to respond slower than the
inner loop controller [23].

3.2 Standard PI-based Field-Oriented Vector Control

A standard PI-based FOC control uses a proportional-integral controller (PI controller) for its speed and current controller. The PI controller is the most commonly used controller in the industry because it only needs two terms to control the system effectively.

\[
\begin{align*}
    u(t) &= K_p e(t) + K_i \int_0^t e(t) dt \\
    &\text{in which } u(t) \text{ is the control output; } e(t) \text{ is the error between desired and measured values; } \ K_p \text{ and } K_i \text{ are a proportional and an integral coefficients, respectively.}
\end{align*}
\]

Figure 3.2 illustrates typical output response of PI controller. Usually, the proportional gain is used to reduce the rise time while the integral gain is used to minimize the steady-state error.

The standard PI-based vector control is comprised of three PI controllers as shown in Fig. 3.3: one PI controller for the speed controller and two PI controllers for the dq current controllers. First, the reference and measured speed are compared each other. Then, the error between two speeds is fed to the speed controller. Second, the speed controller produces the q-axis reference current. The range of q-axis current is from negative to positive maximum allowable current. Generally, the reference d-axis current is zero, but if a flux weakening control or a maximum torque per ampere (MTPA) algorithm is implemented, the d-axis currents vary. The range of the d-axis reference current is between zero and negative maximum allowable current. Usually, the
positive d-axis current is not used because it reduces the resultant torque as shown in Eq. 2-21. Third, these reference currents are fed into the current controller.
The scheme of the inner current-loop controller is developed by rewriting Eq. 2-19 as

\[
\begin{align*}
    v_{sd} &= \left( R_s i_{sd} + L_s \frac{di_{sd}}{dt} \right) - \frac{\omega_e L_{sq} i_{sq}}{\text{comp. term}} \\
    v_{sq} &= \left( R_s i_{sq} + L_s \frac{di_{sq}}{dt} \right) + \frac{\omega_e L_{sq} i_{sq} + \omega_e \lambda_{fd}}{\text{comp. term}}
\end{align*}
\]

(3-2)

in which the item in the parentheses is the output of the current controller while the other item is the compensation term that mitigates the tracking error. The ranges of both references are from negative to positive maximum allowable voltage. The addition of the compensation term with the output of the current controller yields the total dq reference voltage as illustrated in Eq. 3-2.

The next step differs by the type of PWM. If Space Vector PWM (SVPWM) is used, the dq reference voltages are transformed into two-phase \( \alpha \beta \) voltage by using Clarke’s transformation. On the other hand, if Sinusoidal PWM (SPWM) is utilized, the reference voltages are transformed into three-phase \( abc \) voltage by Park’s transformation.

The last step is normalizing the preceding voltage with respect to the PWM type. For SVPWM, the normalized base voltage is \( V_{dc}/\sqrt{3} \) while for SPWM, it is \( V_{dc}/2 \), where \( V_{dc} \) is the DC link voltage. Then, PWM signals insert into the inverter while the inverter changes the current according to the control PWM signal to the motor.

Recent research showed that the standard PI-based vector control experiences the following bottlenecks: a) inevitable output losses due to rise time, overshoot, and settling time, b) hard to obtain the optimal PI coefficients that satisfy both stability and performance, c) no true decoupling between torque and flux.

First, the output response as shown in Fig. 3.2 experiences ineffaceable overshoot, rise time, and settling time. These cause inevitable losses occur in the system, no matter how the coefficients are optimized. Second, there are numerous methods to tune PI gains. However, to
tune PI gains accurately, exact transfer function of actuator is required. Determining the transfer
function for PMSM model is nearly impossible because of non-linearity characteristic of PMSM.
Thus, obtaining the most optimal PI gains for PMSM is nearly impractical. Third, the standard
PI-based control cannot fully decouple the d and q-axis components. To explain this, Eq. 3-2 is
revisited. The equation clearly shows that the control of $v_{sd}$ is mainly come from $i_{sd}$ and has no
major influence on $i_{sq}$, and similar control scheme for $v_{sq}$ is observed. However, this scheme is
inadequate and inaccurate as explained below. If the stator resistance, $R_s$, is neglected, in the
steady-state condition, Eq. 2-19 becomes

$$
I_{sq} = -\frac{v_{sd}}{\omega e L_{sq}}, \quad I_{sd} = \frac{(v_{sq} - \omega e f_d)}{\omega e L_{sd}}
$$  (3-3)

These equations show that the d-axis voltage is mainly effective for $i_{sq}$ or torque control while
the q-axis voltage is primarily effective for d-axis current or field control. However, Eq. 3-3
shows that the intention of the standard vector control is to regulate dq axis currents using $v_{sd}'$
and $v_{sq}'$. In other words, the conventional current controller cannot fully decouple its
compensation term and the main term.

### 3.2.1 Simulink Model

Throughout the thesis, instead of utilizing the inverter and space vector pulse width
modulation (SVPWM) blocks in SimPowerSystem library, the inverter and SVPWM block in
Opal-RT RT-Events are used for two reasons. First, the components in RT-Events produce more
accurate result for discrete simulation of event-based systems than the components in
SimPowerSystem. One of the reasons is because it compensates for the errors introduced when
events occur between samples [26]. Second, it can simulate much faster, for it uses optimized
fixed time-step algorithm. Despite the fact that Simulink offers the fixed time-step simulation,
more lines of code are needed to debug than the code in RT-Events.

The overall conventional motor control schematic is displayed in Fig. 3.4. As illustrated, it comprises of controller block, inverter in Opal-RT RT-Events, and DC source and PMSM model from SimPowerSystem library. First, the controller block receives the reference and actual mechanical speed and three-phase current. The mask view of the controller block is shown in Figure 3.5. Then, through the calculation as illustrated in previous subsection, it produces SVPWM signals. Second, these signals are fed into the inverter block and change DC to three-phase AC. Lastly, the PMSM model produces the actual speed.

![Figure 3.4: Overall motor drive schematic (simulation)](image)
3.3 Neural Network Control

As section 3.2 illustrates, the PI-based vector control experiences few weaknesses. To solve this, this thesis chose an artificial neural network-based vector control over others because it solves all above bottlenecks as follows. First, an artificial neural network is well-known tool to solve non-linear problems by fully incorporating the mathematical equations in training stage. Although the PI-based vector control seems like it is utilizing the mathematical equations to its full extent, it is not apparently. For example, as Eq. 3-2 shows, the current controller of the PI-based control utilizes only compensational terms for mathematical calculation while assuming $V_{sd}'$ and $V_{sq}'$. Unlike the former control, the proposed control fully utilizes the mathematical equations. This incorporation leads to solve the bottlenecks that were exhibited in the PI-based controller.

The sections are organized as follows. First, this section covers the mathematics behind the NN control in motor application. Then, DP and LM+FATT algorithms are discussed.
Examples are included to ease the understanding of the algorithms. Lastly, it provides how the NN control is implemented in Simulink.

### 3.3.1 Mathematical Model

The main objectives of ANN vector control are: 1) to achieve the decoupled d- and q-axis current control, 2) to find the optimal combination of the d- and q-axis control voltages to enhance IPM motor performance, and 3) to make the control system more robust against parameter variations and unknown disturbance such as an impulse of load torque. To develop a current vector controller based ANN, Eq. 2-18 has to be rewritten into the standard state-space form as follows

\[
\frac{d}{dt} \begin{pmatrix} i_{sd} \\ i_{sq} \end{pmatrix} = -\begin{pmatrix} \frac{R_s}{L_d} & -\frac{\omega_L L_q}{L_d} \\ \frac{\omega_L L_d}{L_q} & \frac{R_s}{L_q} \end{pmatrix} \begin{pmatrix} i_{sd} \\ i_{sq} \end{pmatrix} + \begin{pmatrix} \frac{1}{L_d} \\ \frac{1}{L_q} \end{pmatrix} \begin{pmatrix} v_{sd} \\ v_{sq} - \omega_L \lambda_f \end{pmatrix}
\]  

(3-4)

in which \(i_{sd}\) and \(i_{sq}\) are the stator d- and q-axis currents; \(A\) represents the system matrix; \(B\) is the input matrix; and \(u\) signifies the input vector. As illustrated, Eq. 3-4 is in a continuous state-space form. Since the controller will be implemented in a digital controller, the discretization of Eq. 3-4 is required and obtained by utilizing either a zero-order or first-order hold discrete equivalent mechanism as shown by

\[
\begin{pmatrix} i_{sd}(kT_s + T_s) \\ i_{sq}(kT_s + T_s) \end{pmatrix} = A \begin{pmatrix} i_{sd}(kT_s) \\ i_{sq}(kT_s) \end{pmatrix} + B \begin{pmatrix} v_{sd}(kT_s) - 0 \\ v_{sq}(kT_s) - \omega_L \lambda_f \end{pmatrix}
\]

(3-5)

in which \(T_s\) is the sampling period. For \(T_s\) is present on both sides, the equation further simplifies into

\[
i_{sdq}(k + 1) = A \cdot i_{sdq}(k) + B \cdot (v_{sdq}(k) - v_{rdq})
\]

(3-6)
where $k$ is an integer time step; $i_{sdq} = (i_{sd}, i_{sq})'$, $v_{sdq} = (v_{sd}, v_{sq})'$ are the control actions; and $v_{rdq} = (0, \omega_r, 0)'$ illustrates the induced voltage of the rotor permanent magnet.

Figure 3.6 illustrates overall motor drives with neural network current controller and neural network structure. As it shows, the proposed neural network is structured into three different layers: an input layer, an action hidden network layer, and an output layer. The main role of the input layer is taking the error terms and the integrals of the error terms, $e_{dq}$ and $s_{dq}$, and then, transform those by dividing the appropriate gains and applying the hyperbolic tangents. The error and integrals of the error terms are expressed as

$$
e_{dq}(k) = i_{dq}(k) - i_{dq}^*(k), \quad s_{dq}(k) = \int_0^k e_{dq}(k) \, dt \quad (3-7)$$

Then, these outputs of the input layer feed forward to action network layer. This layer is comprised of two hidden layers of six nodes. Like in the input layer, each node is computed using hyperbolic tangent functions. The last layer is called output layer. The primary objective of
the output layer is translating the outputs of the action network layer into the reference dq
voltages. This was achieved by applying the hyperbolic tangent. Then, because these outputs are
normalized, a gain has to be multiplied. This gain is equaled to the maximum allowable voltage,
which is depended on amplitude of DC voltage and type of PWM. For example, since SVPWM
was used, the maximum allowable voltage is

\[ k_{\text{supwm}} = V_{dc} \cdot \sqrt{3}/2 / \sqrt{3} \]  

(3-8)

As a result, the final control action \( v_{sdq} \) is defined as

\[ v_{sdq}(k) = k_{\text{supwm}} \cdot A(e_{dq}(k), s_{dq}(k), w) \]  

(3-9)

where \( w \) is an weight vector while \( A(\cdot) \) represents the action neural network.

Note that RNN is used for training while FNN is used when it is implemented in vector
control.

3.3.2 Training

Recently, many researches had been conducted in the area of dynamic programming (DP)
for nonlinear and complex systems. Adaptive critic designs (ACD) employ approximate dynamic
programming methods to determine the optimal cost and the control of a system [25]. Further,
dynamic programming was also used to control a turbogenerator [26]. In addition, DP along with
backpropagation through time (BPTT) further is implemented in recurrent neural networks
(RNNs), where BPTT was combined with Resilient Propagation (RPROP) to stimulate the
convergence of training. For this application, either PID or predictive control is employed [27].
This integration caused to produce significant advantages, such as stable control under error-
prone physical system, zero steady-state error, and other. However, the studies show that such
training technique experience few issues, including slow convergence and oscillation problems
that cause training to diverge. Thereby, this thesis utilizes different training techniques, called Levenberg-Marquardt and forward accumulation through time (LM+FATT), along with DP not only to fulfill the same performances, but also accelerate the training.

### 3.3.2.1 Dynamic Programming

Dynamic programming (DP) is a very powerful algorithmic technique that finds optimal solution by determining the minimum set in the system. According to [28], DP often is used to solve optimization problems that need sequences of decisions by determining the solutions for each iteration. To clarify the concept of the DP more, an example is presented in this section.

Figure 3.7 illustrates the DP example with the cost and node associated with it. The objective of this problem is to find the minimum cost route from A to N [29]. For minimum cost route, it is called the optimal policy; any other subsequence is a sub-policy. Further, as the figure

![Figure 3.7: DP example](image)
shows, there are total of five stages (I, II, III, IV, and V) and six nodes \( \{X_i\} \). The symbol \( V_a(X_i, X_{i+1}) \) represents the cost of traveling stage, whereas \( f_a(X_i) \) denotes the minimum cost for each stage \( a \) and node \( i \). Six nodes are expressed as

\[
\{X_0\}: A \quad \{X_2\}: E, F, G \quad \{X_4\}: L, M \\
\{X_1\}: B, C, D \quad \{X_3\}: H, I, J, K \quad \{X_5\}: N
\]

The minimum cost for the first stage is

\[
f_1(B) = V_1(A, B) = 5 \quad (3-10)
\]

\[
f_1(C) = V_1(A, C) = 2 \quad (3-11)
\]

\[
f_1(D) = V_1(A, D) = 3 \quad (3-12)
\]

The minimum cost for the first and second stage is

\[
f_{II}(E) = \min[f_1(X_1) + V_{II}(X_1, E)] = \min\left[\frac{5 + 11}{B}, \frac{2 + 8}{C}, \frac{3 + \infty}{D}\right] = 10 \quad (3-13)
\]

Since D cannot reach E directly, the value of infinite is used to show that there is no connection.

Thus, the cost of F and G becomes

\[
f_{II}(F) = \min[f_1(X_1) + V_{II}(X_1, F)] = \min[\infty, 6, 9] = 6 \quad (3-14)
\]

\[
f_{II}(G) = \min[f_1(X_1) + V_{II}(X_1, G)] = \min[\infty, 11, 9] = 9 \quad (3-15)
\]

Likewise, the minimum costs for other stages are calculated. For example, the minimum cost of stages I through IV as a function of \( X_4 \) is

\[
f_{IV}(X_4) = \min[f_{III}(X_3) + V_{IV}(X_3, X_4)]
\]

\[
f_{IV}(L) = \min[13 + 9, 12 + 3, 11 + 7, 13 + \infty] = 15, X_3 = H = I = J = K \quad (3-17)
\]

\[
f_{IV}(M) = \min[13 + \infty, 12 + 6, 11 + 8, 135] = 18, X_3 = H = I = J = K \quad (3-18)
\]

Then, the minimum path can be achieved by tracing back the previous minimum cost calculation.

As a result, it becomes
<table>
<thead>
<tr>
<th>Stage</th>
<th>({X_i})</th>
<th>(f_i)</th>
</tr>
</thead>
<tbody>
<tr>
<td>1</td>
<td>B, C, D</td>
<td>5, 2, 3</td>
</tr>
<tr>
<td>2</td>
<td>E, F, G</td>
<td>10, 6, 9</td>
</tr>
<tr>
<td>3</td>
<td>H, I, J, K</td>
<td>13, 12, 11, 13</td>
</tr>
<tr>
<td>4</td>
<td>L, M</td>
<td>15, 18</td>
</tr>
<tr>
<td>5</td>
<td>N</td>
<td>19</td>
</tr>
</tbody>
</table>

Table 3.1: Results of DP example

where bold letters and numbers are the example’s optimal path [29].

The DP cost function for NN-based PMSM control is

\[
C (i_{dq}(j)) = \sum_{k=j}^{\infty} \gamma^{k-j} U \left( e_{dq}(k) \right), j > 0, 0 < \gamma \leq 1
\]  

(3-19)

where \(\gamma\) is a discount factor and \(U(\cdot)\) the local cost or utility function. The objective of DP is to minimize the error of the dq current. The utility function is defined as

\[
U(e_{dq}(k)) = \left[ e_d^2(k) + e_q^2(k) \right]^\alpha
\]

\[
= \left\{ \left[ i_d(k) - i_{dref}(k) \right]^2 + \left[ i_q(k) - i_{qref}(k) \right]^2 \right\}^\alpha
\]  

(3-20)

in which \(\alpha\) is 1 for the motor drive application.

3.3.2.2 Levenberg-Marquardt Algorithm

A LM algorithm is a powerful tool that is used to solve non-linear least squares minimization [30]. Usual non-linear function is of the following special form

\[
f(x) = \frac{1}{2} \sum_{j=1}^{m} r_j^2(x)
\]  

(3-21)

where \(x\) is comprised of \(n\) vectors while \(r_j\) is the function of \(x\). Like other minimization method, the LM algorithm also uses iterative technique to find the minimized value. To accomplish this task, the LM interpolates between the steepest descent method and the Gauss-Newton algorithm. First, the LM uses the steepest descent method when the desired value is far from a threshold value. The objective of this method is to bring the desired value near to a threshold value. Even
though this method is slow, it guarantees convergence of the value. When the desired value is near to a threshold value, the LM behaves like Gauss-Newton method. The main objective of this method is to find the most optimal minimized value. To clarify the concept more, this section will recapitulate the backgrounds of the LM.

Let’s assume that the function that needs to be minimized is called $S(\beta)$ while $(x_i, y_i)$ is independent and dependent variables that optimize the parameters $\beta$ of the curve $f(x_i, \beta)$.

$$S(\beta) = \sum_{i=1}^{m} [y_i - f(x_i, \beta)]^2$$  \hspace{1cm} (3-22)

In each iteration, the term $\beta$ is replaced by a new estimate, $\beta + \delta_p$. The term $\delta_p$ is determined by using following function.

$$f(x_i, \beta + \delta_p) \approx f(x_i, \beta) + J_i \delta_p, \text{ where } J_i = \frac{\partial f(x_i, \beta)}{\partial \beta}$$  \hspace{1cm} (3-23)

where $J_i$ is the gradient of the curve, $f$ with respect to the parameter, $\beta$. When the function $S(\beta)$ is near to its minimum, the gradient of the function with respect to $\delta_p$ will be zero. This yields the function to be rewritten as

$$S(\beta + \delta_p) \approx \sum_{i=1}^{m} (y_i - f(x_i, \beta) - J_i \delta_p)^2$$  \hspace{1cm} (3-24)

If the right side is differentiated with respect to $\delta_p$ and the left side is assumed to be zero, the equation becomes

$$(J^T J) \delta_p = J^T [y - f(\beta)]$$  \hspace{1cm} (3-25)

in which $J$ is the Jacobian matrix and $y$ are vectors with $i^{th}$ component. To make this more flexible, Levenberg inserted the identity matrix, $I$, and non-negative damping factor, $\lambda$.

$$(J^T J + \lambda I) \delta_p = J^T [y - f(\beta)]$$  \hspace{1cm} (3-26)

The damping factor is adjusted at each iteration and determines which algorithm will be used. If the rate of reduction of the cost function is fast, the LM algorithm will behave like Gauss-
Newton algorithm by using small $\lambda$, whereas if it is slow, the LM algorithm will mimic the behavior of gradient descent algorithm by using large $\lambda$.

To apply the LM in the PM motor application, the first step is rearranging the cost function, $C(\cdot)$ (Eq. 3-19), into the sum-of-squares (Eq. 3-21). If the $\gamma$ and $j$ are 1 while $k$ is the positive integer, the equation becomes

$$
C = \sum_{k=1}^{N} U(e_{dq}(k)) \iff \text{define } V(k) = \sqrt{U(e_{dq}(k))} \implies C = \sum_{k=1}^{N} (V(k))^2 \quad (3-27)
$$

And the gradient of the cost function with respect to the weight vector is

$$
\frac{\partial C}{\partial \vec{w}} = \frac{\partial \sum_{k=1}^{N} (V(k))^2}{\partial \vec{w}} = \sum_{k=1}^{N} 2V(k) \frac{\partial V(k)}{\partial \vec{w}} = 2J_p(\vec{w})^T V \quad (3-28)
$$

in which the Jacobian matrix and the error function are defined as

$$
J_p(\vec{w}) = \begin{bmatrix}
\frac{\partial V(1)}{\partial w_1} & \cdots & \frac{\partial V(1)}{\partial w_M} \\
\vdots & \ddots & \vdots \\
\frac{\partial V(N)}{\partial w_1} & \cdots & \frac{\partial V(N)}{\partial w_M}
\end{bmatrix}, \quad V = \begin{bmatrix} V(1) \\ \vdots \\ V(N) \end{bmatrix} \quad (3-29)
$$

where the weight update is expressed as

$$
\Delta \vec{w} = -[J_p(\vec{w})^T J_p(\vec{w}) + \mu I]^{-1} J_p(\vec{w})^T V \quad (3-30)
$$

### 3.3.2.3 Forward Accumulation Through Time Algorithm

Eq. 3-30 clearly shows that Jacobian matrix is the core term to be determined. Among other method that finds Jacobian matrix, a Forward Accumulation Through Time (FATT) was chosen because it incorporates the procedures of determining the derivatives of the Jacobian matrix and the DP cost into one system for each training epoch and unrolling of the system trajectory [31].
The first procedure is finding the \( k \)th row of the Jacobian matrix \( J_p(w) \) by taking the derivative of the error function \( V \) respect to the weight vector.

\[
\frac{\partial V(k)}{\partial w} = \frac{\partial V(k)}{\partial e_{dq}(k)} \cdot \frac{\partial e_{dq}(k)}{\partial w}
\]

where the first term is equal to

\[
\frac{\partial V(k)}{\partial e_{dq}(k)} = \frac{\partial}{\partial e_{dq}(k)} \left[ \left( e_d^2(k) + e_q^2(k) \right)^{1/2} \right] = \left( e_d^2(k) + e_q^2(k) \right)^{-1/2} \cdot \left[ e_d(k) e_q(k) \right]
\]

And the second term is

\[
\frac{\partial e_{dq}(k)}{\partial w} = \frac{\partial l_{dq}(k)}{\partial w}
\]

The derivative, \( \frac{\partial l_{dq}(k+1)}{\partial \bar{w}} \), which is for the next time period, is illustrated as

\[
\frac{\partial l_{dq}(k+1)}{\partial \bar{w}} = A \frac{\partial l_{dq}(k)}{\partial \bar{w}} + B \frac{\partial v_{rdq}(k)}{\partial \bar{w}}
\]

Since the last term \( \frac{\partial v_{rdq}}{\partial \bar{w}} \) is constant, the second term becomes negligible. Finally, the term \( \frac{\partial v_{rdq}(k)}{\partial \bar{w}} \) is equal to

\[
\frac{\partial v_{rdq}(k)}{\partial \bar{w}} = k_{PWM} \left( \frac{\partial A(k)}{\partial e_{dq}(k)} \cdot \frac{\partial l_{dq}(k)}{\partial \bar{w}} + \frac{\partial A(k)}{\partial s_{dq}(k)} \cdot \frac{\partial s_{dq}(k)}{\partial \bar{w}} + \frac{\partial A(k)}{\partial \bar{w}} \right)
\]

The aforementioned equations are combined into algorithm. In the algorithm, \( \phi(k) = \sum_{j=1}^{J} I_q(j) \) and

\[
\frac{\partial \phi(k)}{\partial \bar{w}} = \sum_{j=1}^{J} \frac{\partial I_q(j)}{\partial \bar{w}} \cdot \frac{\partial \phi(k)}{\partial \bar{w}} = \frac{\partial \phi(k)}{\partial \bar{w}}
\]

**Algorithm 1:** FATT algorithm to calculate the Jacobian matrix.

1: \( C \leftarrow 0, \bar{e}_{dq}(0) \leftarrow 0, \bar{s}_{dq}(0) \leftarrow 0, \frac{\partial \bar{e}_{dq}(0)}{\partial \bar{w}} \leftarrow 0, \frac{\partial \bar{e}_{dq}(0)}{\partial \bar{w}} \leftarrow 0 \)

2: \{Calculate Jacobian matrix \( J_p(w) \}\}

3: for \( k = 0 \) to \( N-1 \) do

4: \( \bar{v}_{rdq}(k) \leftarrow k_{PWM} A(e_{dq}(k), x_{dq}(k), \bar{w}) \)

5: \( \frac{\partial \bar{v}_{rdq}(k)}{\partial \bar{w}} \leftarrow T_e \left( \frac{\partial \phi(k)}{\partial \bar{w}} - \frac{1}{2} \frac{\partial \bar{l}_{dq}(k)}{\partial \bar{w}} \right) \)
6: \[ \frac{\partial \tilde{v}_{dq}(k)}{\partial w} \leftarrow k_{p_{\omega}} \left( \frac{\partial \tilde{e}_{dq}(k)}{\partial w} + \frac{\partial \tilde{e}_{dq}(k)}{\partial \tilde{e}_{dq}(k)} \frac{\partial \tilde{e}_{dq}(k)}{\partial \tilde{e}_{dq}(k)} + \frac{\partial \tilde{A}(k)}{\partial \tilde{w}} \right) \]

7: \[ \frac{\partial \tilde{i}_{dq}(k+1)}{\partial w} \leftarrow A \frac{\partial \tilde{i}_{dq}(k)}{\partial w} + B \frac{\partial \tilde{v}_{dq}(k)}{\partial w} \]

8: \[ \tilde{i}_{dq}(k+1) \leftarrow A \tilde{i}_{dq}(k) + B (\tilde{v}_{dq}(k) - \tilde{v}_{dq}) \]

9: \[ e_{dq}(k+1) \leftarrow \tilde{i}_{dq}(k+1) - \tilde{i}_{dq_{ref}}(k+1) \]

10: \[ s_{dq}(k+1) \leftarrow s_{dq}(k) + \frac{T_e}{2} (e_{dq}(k+1) + e_{dq}(k)) \]

11: \[ C \leftarrow C + U (\tilde{e}_{dq}(k+1)) \]

12: \[ \frac{\partial \tilde{r}(k+1)}{\partial w} \leftarrow \frac{\partial \tilde{r}(k)}{\partial w} + \frac{\partial \tilde{i}_{dq}(k+1)}{\partial \tilde{w}} \]

13: \[ \frac{\partial V(k+1)}{\partial \tilde{w}} \leftarrow \frac{\partial V(k+1)}{\partial \tilde{e}_{dq}(k+1)} \]

14: \[ \text{the}(k+1)\text{th row of } J_{\tilde{w}} \leftarrow \frac{\partial V(k+1)}{\partial \tilde{w}} \]

15: \[ \text{end for} \]

16: \{on exit, the Jacobian matrix \( J_{\tilde{w}} \) is finished for the whole trajectory\}

Table 3.2: FATT algorithm to calculate the Jacobian matrix

3.3.2.4 Training Algorithm

The overall training flow chart is illustrated in Figure 3.8. The parameters in the flow chart are explained as follows

- \( \mu_{\text{max}} \): maximum available and acceptable \( \mu \)
- \( \beta_{\text{de}} \): decreasing factor (adjust the learning rate during the training)
- \( \beta_{\text{in}} \): increasing factor (adjust the learning rate during the training)
- \( \text{Epoch}_{\text{max}} \): maximum number of training epochs
- \( \| \partial C/\partial \tilde{w} \|_{\text{min}} \): norm of the minimum acceptable gradient

The final objective of this process is minimizing the DP cost by manipulating \( \mu \). The weights are calculated by using the Cholesky factorization [31], which solves the linear equation twice as efficient as the LU decomposition. If the following requirements are met, the process will end: a)
training epoch = \( Epoch_{\text{max}} \), b) \( \mu > \mu_{\text{max}} \), and c) \( \| \frac{\partial C}{\partial \tilde{w}} \| < \| \frac{\partial C}{\partial \tilde{w}} \|_{\text{min}} \). As Figure 3.9 illustrates, the DP cost of the proposed NN controller was stabilized within 100 iterations. The time that took to train the controller was less than 20 minutes.

![Overall flowchart of training NN controller using LM+FATT algorithm](image)
3.3.3 Simulink Model

The overall schematic is same as shown in Fig. 3.6. The major difference between the conventional and the proposed NN control is the structure of the current controller. Figure 3.10 shows the block diagram of the NN control while Figure 3.11 shows the mask view of the current controller. Figure 3.12 illustrates how the aforementioned RNN is translated to Simulink block.

Figure 3.10: Block diagram of the NN controller
Figure 3.11: Block diagram of the NN current controller

Figure 3.12: Block diagram representation of RNN control
 CHAPTER 4
MAXIMUM TORQUE PER AMPERE AND FLUX WEAKENING

Generally, there are two operating regions in PMSM drives: constant torque and constant power region. Figure 4.1 shows a typical torque versus speed graph with two operating regions. First, when the motor speed is between 0 and its base speed, a PMSM is in constant torque region. In this region, a PMSM can produce the torque from 0 to its maximum torque. Because the maximum torque can be sustained constantly in this region, it is called constant torque region. When the motor speed exceeds its base speed, which occurs when the back-EMF of the motor reaches the maximum voltage of the system, the maximum allowable torque starts to decrease to prevent the back-EMF to be exceeded. As the below equation shows, the electrical and mechanical powers are interrelated to each other.

\[ P = VI = T \omega \] (4-1)

where \( T \) is the motor torque and \( \omega \) is the motor speed. Since the power is constant as shown in green curve in Fig. 4.1, this region is called constant power region. In the constant torque region, many studies utilize a control called maximum torque per ampere (MTPA) control because it finds the most optimal combination of dq currents to produce the demanded torque [32]. This technique will not only minimize the copper loss, but also improve the overall driving distance, since less current is consumed.

In the constant power region, a control called flux weakening control is used. This control sacrifices the current to increase a motor’s speed range by producing the flux against the flux that
is produced by the magnet. Since the induced voltage is proportional to the rate of the flux respect to time, if the amplitude of the flux is less, the lesser induced voltage will be induced on the stator. This control is very important because usually, without this control in HEV application, the maximum vehicle speed is around 50 or 60 mph.

Note that MTPA and FW controls are operating before the proposed NN control. The main objective is translating the reference torque into optimal d- and q-axis currents for MTPA control and into adequate d- and q axis currents for FW control.

Figure 4.1: Torque versus speed curve of PMSM with other descriptions

The section is divided into three sections. The first subsection covers the general overview of two controls while the next two sections epitomize the MTPA control scheme and the flux weakening control, respectively.

4.1 General operation of PMSM

As Figure 4.2a shows, the motor operation is divided into four sections. As the figure illustrates, four sections are divided by three different speeds, $\omega_1$ to $\omega_3$, which are called a base
speed, a rated speed, and a second-breakdown speed, respectively. Fig 4.2b illustrates the four sections in $i_d$, $i_q$ current contour. This subsection will cover how these sections are divided.

Figure 4.2: a) Torque and speed curve with four different regions b) Segmentation of operation region for current contour region [32]
The first section occurs between 0 and $\omega_1$, which is a base speed. The concept of the base speed begins with voltage-limit ellipse and current-limit circle. Usually, the inverter limits the current rating while the battery determines the voltage rating. The mathematical representations of these ratings are

$$U_s = \sqrt{U_d^2 + U_q^2} \leq U_{max}$$  \hspace{1cm} (4-2)

$$I_s = \sqrt{I_d^2 + I_q^2} \leq I_{max}$$  \hspace{1cm} (4-3)

where $I_s$ and $U_s$ are the amplitudes of operating current and voltage, respectively. If ohmic voltage drops across the stator winding are neglected in Eq. 2-18, Eq. 4-2 can be rewritten in terms of the motor parameter

$$\sqrt{U_d^2 + U_q^2} = \sqrt{(\omega_e L_q i_{sq})^2 + (\omega_e L_d i_{sd} + \lambda_{f} \omega_e)^2} \leq U_{max}$$  \hspace{1cm} (4-4)

The concept is more clarified if Eq. 4-4 is rewritten in an ellipse equation.

$$\left(\frac{x}{a}\right)^2 + \left(\frac{y}{b}\right)^2 = \left(\frac{i_d}{\frac{U_{max}}{L_d \omega_e}}\right)^2 + \left(\frac{i_q}{\frac{U_{max}}{L_q \omega_e}}\right)^2 = 1$$  \hspace{1cm} (4-5)

where $a$ and $b$ are the length of the semi-major and semi-minor axis, respectively, as shown in Figure 4.3. Eq. 4-5 clearly shows that if $a$ and $b$ are decreased, the ellipse will shrink. Likewise, if the speed is increased, $U_{max}/L_d \omega_e$ and $U_{max}/L_q \omega_e$ are decreased; then, the ellipse will shrink.

By using Eq. 4-4, the base speed can be calculated. To simplify the calculation, the inductances and flux linkage constant are assumed to be constant. However, in reality, these values alter as the motor operates. If the motor parameters are constant, the terms that are changing in Eq. 4-4 are $d$- and $q$-axis currents and the speed. If the speed is below the base speed, the value of $d$- and $q$-axis currents can be found by using MTPA method while if the speed is above, different calculation methods have to be used. These methods will be covered in next subsection. Before
that, Figure 4.4 illustrates a typical curve of MTPA control with voltage-limit and current-limit ellipse in $i_d, i_q$ plane.

Figure 4.3: Ellipse

Figure 4.4: Voltage and current limit circle with constant torque curve and MTPA for IPMSM
If the speed exceeds the base speed, the motor operation is in the second section. This section operates between the base speed and $\omega_2$, which is a rated speed. The rated speed is a speed where voltage-limit ellipse, current-limit circle, and constant torque curve meet. When the motor is in this region, MTPA control cannot be used. Instead, the reference torque moves along the voltage-limit ellipse within the current-limit circle to prevent the back-EMF to exceed the maximum voltage. Therefore, optimal control cannot be used. Figure 4.2b shows the aforementioned behavior in $i_d$, $i_q$ plane.

When the motor speed exceeds the rated speed, the motor now operates in third section. In third section, as Fig. 4.2b shows, it moves along the current-limit circle. Like in the second section, optimal control cannot be used.

If the motor speed exceeds a second-breakdown speed, the motor operates in last section. In this region, the reference torque curve intersects the voltage limit tangentially; thus, different method needs to be used. As Fig. 4.2b shows, a maximum torque per flux line is used for the reference torque to be followed.

### 4.2 MTPA Algorithm

Among many published controls, this thesis chose MTPA control because 1) it uses both d-and q-axis currents for control 2) minimizes the copper loss, which is one of dominant losses in the motor [32], and 3) quite simple relative to other methods. In this thesis, three MTPA controls of IPMSM are covered.

The concept of MTPA control begins with the stator dq voltage equations of PMSMs (Eq. 4-6) and the electromagnetic torque equation (Eq. 4-7). The detailed explanations of these equations were covered in Chapter 2.
\[
\begin{bmatrix}
    v_{sd} \\
    v_{sq}
\end{bmatrix} = \begin{bmatrix}
    R_s & -\omega_e L_d \\
    \omega_e L_d & R_s
\end{bmatrix} \begin{bmatrix}
    i_{sd} \\
    i_{sq}
\end{bmatrix} + \begin{bmatrix}
    0 \\
    \lambda_{fd} \omega_e
\end{bmatrix}
\]

(4-6)

\[
T_{em} = \frac{p}{2} (\lambda_{fd} i_{sq} + (L_{sd} - L_{sq}) i_{sd} i_{sq})
\]

(4-7)

where \(v_{sdq}\) is the stator dq voltage; \(R_s\) is the stator resistance; \(\omega_e\) is the electrical angular velocity; \(L_{dq}\) is the stator dq inductance; \(\lambda_{fd}\) is the flux linkage constant; \(T_{em}\) is the electromagnetic torque; and \(p\) is the pole pairs. If the motor is SPMSM, the electrodynamic torque equation is only dependent on the q-axis current because the stator dq inductances are almost equal to each other; thus, the reluctance torque, which is the second portion of Eq. 4-7, is ignored. Thereby, the electromagnetic torque of SPMSM can be written as

\[
T_{em} = \frac{p}{2} (\lambda_{fd} i_{sq})
\]

(4-8)

Unlike SPM machine, IPM machine has different d- and q-axis inductances. This means that the second portion in Eq. 4-7 cannot be ignored; thus, the d-axis current has to be included in the calculation to control the motor efficiently.

A MTPA control for SPM machine is very easy, since the electromagnetic torque is only dependent on the q-axis current, as shown in Eq. 4-8. On the other hands, implementing a MTPA control for IPM machine is hard because the electromagnetic equation is dependent on both d- and q-axis currents. Thus, many literatures are still researching MTPA controls for IPM machine. Among published literatures, three MTPA controls are covered as follows.

The first MTPA control is based on look-up table, which is widely used in many applications because of simplicity. The values of d- and q-axis current are calculated on the basis of the required torque and speed \textit{a priori}. However, it will not provide accurate result because, generally, constant motor parameters are used to calculate the data. If variable motor parameters are included in the calculation, the dataset will be too large to be implemented.
In lieu of look-up table method, the second control is based on real-time calculation. Most studies use these equations to calculate the d- and q-axis current [40] because of simplicity.

\[
i_d = \frac{-\lambda_f d + \sqrt{\lambda_f d^2 + b(L_d - L_q)^2}}{2(L_d - L_q)} \tag{4-9}
\]

\[
i_q = \sqrt{I_s^2 - i_d^2} \tag{4-10}
\]

where \(I_s\) is the reference current that changes. Usually, a PI controller is used to produce the reference current. However, this control is only suitable when the reference input is speed, but not suitable for applications that require reference torque as input. Generally, the motor controller in EV uses the reference torque as its input, which is produced by a controller called supervisory controller. Thus, this control will not covered in details.

The third method covers above weaknesses by using Ferrari’s method. This method receives the torque as its reference and converts to corresponding d- and q-axis currents by solving quartic equations in real time [32]. This also finds the optimal value by finding the minimum value. In constant torque region, the constrained minimization problem is formulated as

Minimize \(i_d^2 + i_q^2\)

subject to \(\frac{3p}{2}(\lambda_f d i_q + (L_d - L_q)i_d i_q) = T_{\text{demand}}\)

The Lagrangian for this problem is formulated as

\[
L(i_d, i_q, \lambda) = i_d^2 + i_q^2 + \lambda \left( \frac{3p}{2}(\lambda_f d i_q + (L_d - L_q)i_d i_q) - T_{\text{demand}} \right) \tag{4-11}
\]

in which \(\lambda\) is a Lagrangian multiplier. Then, the conditions for minimization are defined as

\[
0 = \frac{\partial L}{\partial i_d} = 2i_d + \frac{3p}{2} \lambda (L_d - L_q) i_q \tag{4-12}
\]

\[
0 = \frac{\partial L}{\partial i_q} = 2i_q + \frac{3p}{2} \lambda (\lambda_f d + (L_d - L_q)i_q) \tag{4-13}
\]
\[ 0 = \frac{\partial L}{\partial \lambda} = \frac{3p}{2} \left( \lambda_{fd} i_q + (L_d - L_q) i_d i_q \right) - T_{demand} \tag{4-14} \]

Then, the quartic polynomial equations of the IPM machine is

\[ i_d^4 + A_1 i_d^3 + B_1 i_d^2 + C_1 i_d + D_1 = 0 \tag{4-15} \]

where

\[ A_1 = \frac{3\lambda_{fd}}{(L_d - L_q)}, \quad B_1 = \frac{3\lambda_{fd}^2}{(L_d - L_q)^2}, \quad C_1 = \frac{\lambda_{fd}^3}{(L_d - L_q)^3}, \quad \text{and} \quad D_1 = -\frac{167r_{d}^2}{9(p/2)^2\left(L_d - L_q\right)^2}. \]

The general solution is given as

\[ i_d = \frac{\left(-p_1 - \sqrt{p_1^2 - 8*p*Q1}\right)}{4} \tag{4-16} \]

\[ i_q = \frac{\left(\frac{2}{3}\right) T_{demand}}{p} \tag{4-17} \]

\[ P1 = A_1 + \sqrt{A_1^2 - 4*B_1 + 4*Y_1} \tag{4-18} \]

\[ Q1 = Y_1 - \sqrt{Y_1^2 - 4*D_1} \tag{4-19} \]

\[ Y_1 = S_1 + T_1 - \frac{Car_{B1}}{3} \tag{4-20} \]

\[ S_1 = \left(R_1 + \sqrt{Q2^3 + R_1^2}\right)\frac{1}{3} \tag{4-21} \]

\[ T_1 = -\left|R_1 - \sqrt{Q2^3 + R_1^2}\right|\frac{1}{3} \tag{4-22} \]

\[ Q2 = \frac{3*Car_{A1}*Car_{C1} - Car_{B1}}{9*Car_{A1}^2} \tag{4-23} \]

\[ R_1 = \frac{9*Car_{A1}*Car_{B1}*Car_{C1} - 27*Car_{A1}^2*Car_{D1} - 2*Car_{B1}^3}{54*Car_{A1}^3} \tag{4-24} \]

\[ Car_{A1} = 1, \quad Car_{B1} = -B_1, \quad Car_{C1} = A_1 * C_1 - 4 * D_1, \]

\[ Car_{D1} = 4 * B_1 * D_1 - A_1^2 * D_1 - C_1^2 \tag{4-25} \]

4.3 Flux Weakening Algorithm

When the motor speed exceeds the base speed, the flux weakening control is needed to
prevent the back-EMF to exceed the maximum allowable voltage. In this thesis, real-time flux weakening controls are focused rather than \textit{a priori} method because determining exact motor parameters in high speed is difficult.

4.3.1 PI-Based Control

Figure 4.5 shows overall control scheme of the Pi-based control while Figure 4.6 illustrates the C-code in Switch block.

\begin{figure}[h]
\centering
\includegraphics[width=0.8\textwidth]{figure4.5.png}
\caption{Adaptive flux weakening control method}
\end{figure}

\begin{verbatim}
if (I_{d1}^* != 0 && U_m \leq 0.95*U_{max})
    Output = CompTerm;
else
    Output = e_m;
\end{verbatim}

\begin{figure}[h]
\centering
\includegraphics[width=0.8\textwidth]{figure4.6.png}
\caption{C-code in Switch block}
\end{figure}

where $U_{max}$ is the maximum allowable voltage, which is $V_{dc}/\sqrt{3}$ for SVPWM; $\text{CompTerm}$ is the compensational term; $I_{d1}^*$ is the d-axis current before saturation; $I_d^*$ is the real d-axis current; and $U_m$ is the actual magnitude of the voltage, which can be calculated by

$$U_m = \sqrt{U_d^2 + U_a^2} \quad (4-26)$$

First, this value is filtered at time constant of 1 ms using low-pass filter to remove the noise, and then, it is compared with the maximum allowable voltage. The preceding error inserts
into Switch block. The Switch block determines corresponding output by using the code in Fig. 4.6, and then, this output is inserted to the a PI controller. The output of the PI controller is saturated between 0 and negative infinity to pass only the negative value. Finally, the maximum q-axis current is computed to prevent over-current. For PI gains, 20 and 200 were used, respectively.

4.3.2 Hysteresis Discrete Control

The origin of this control is from a bang-bang controller, also known as a hysteresis controller, which is a feedback controller that switches between two known states. This controller is implemented to calculate the d-axis current by using below equation.

\[ i_d^*(k) = i_d^*(k - 1) - 0.1 \left( (U_{ref} - U_{max}) \times 100 \right)^2 \]  (4-27)

where \( i_d(k-1) \) is the past value of \( i_d(k) \). The operation of this controller is simple. If the reference voltage exceeds the maximum, the second portion decreases the current value with respect to the error between reference and maximum voltage. However, the downside of this controller is that like other hysteresis controller, the d-axis current changes in sawtooth waveform as shown in Fig. 4.6.

4.3.3 Ferrari’s Method Control

Ferrari’s method is used to determine the d-axis current in flux weakening region. Theoretically, this control ensures to get optimal d- and q-axis currents for different constraints. However, the downside of this method and other optimal method is that it requires accurate parameter values to determine reference d- and q-axis current. Another downside of this control is that it produces complex numbers. The complex number was found when square root of Q3
and \( R_2 \) is in process for \( S_2 \) and \( T_2 \). Due to that, the program halts. The quartic equation in flux weakening region is expressed as [32].

\[
i_d^4 + A_2 i_d^3 + B_2 i_d^2 + C_2 i_d + D_2 = 0 \quad (4-28)
\]

where

\[
A_2 = \frac{2\lambda_{fd}}{L_d - L_q} \left( 2 - \frac{L_q}{L_d} \right), \quad B_2 = \frac{\lambda_{fd}^2}{L_d (L_d - L_q)} + \frac{4\lambda_{fd}^2}{L_d (L_d - L_q)} - \frac{\lambda_{fd}^2}{\omega_e^2 L_d^2}, \quad C_2 = \left( \frac{2\lambda_{fd}}{\omega_e^2} \right) \left( \frac{\lambda_{fd}^2}{L_d (L_d - L_q)} \right)
\]

\[
D_2 = \frac{1}{(L_d - L_q)^2} \left( \frac{\lambda_{fd}^4}{L_d^2} + \frac{L_q^2}{L_d^2} \cdot \frac{16\tau_{T_d}^{\text{demand}}}{9} - \frac{V_e^2}{\omega_e^2} \frac{\lambda_{fd}^2}{L_d^2} \right).
\]

where \( i_d \) and \( i_q \) can be determined using below equations.

\[
i_d = \frac{-p_2 + \sqrt{p_2^2 - 8Q_3}}{4} \quad (4-29)
\]

\[
i_q = \frac{\left( \frac{3}{2} \right)^p \cdot \frac{T_{\text{demand}}}{\lambda_{fd} + (L_d - L_q) \cdot i_d}}{P} \quad (4-30)
\]

\[
S_2 = S_2 + T_2 = \frac{Car_{B2}}{3} \quad (4-31)
\]

\[
R_2 = \frac{9 \cdot Car_{A2} \cdot Car_{B2} \cdot Car_{C2} - 27 \cdot Car_{A2} \cdot Car_{D2} - 2 \cdot Car_{B2}}{54 \cdot Car_{A2}^3} \quad (4-34)
\]

\[
Q_3 = \frac{3 \cdot Car_{A2} \cdot Car_{C2} \cdot Car_{B2}^2}{9 \cdot Car_{A2}^2} \quad (4-35)
\]

\[
Car_{A2} = 1, Car_{B2} = -B_2, Car_{C2} = A_2 \cdot C_2 - 4D_2, Car_{D2} = 4B_2D_2 - A_2^2D_2 - C_2^2 \quad (4-36)
\]

### 4.3.4 Constant Voltage Constant Power (CVCP) Control

The simplest flux weakening is presented as below [33]. It is based on keeping the voltage and power constant as shown in below equations.

\[
P_m = T_m \omega = \frac{3}{2} (v_d i_d + v_q i_q) = \text{Const.} \quad (4-37)
\]
\[ v_d \approx -\omega_e L_q i_q = -\omega_{base} L_q i_{qb} = \text{Const.} \]  (4-38)

\[ v_q \approx \lambda_{fd} \omega_e + \omega_e L_d i_d = \lambda_{fd} \omega_{base} = \text{Const.} \]  (4-39)

\[ v_s = V_{s_{max}} = V_{qb} = \text{const.} \]  (4-40)

where \( v_{qb}, i_{qb}, \) and \( \omega_{base} \) are base voltage, base current, and base speed, respectively. For SPM motor case, the base current is the maximum current while for IPM motor application, the base current differs. If Eq. 4-37 and 4-40 are combined, the d and q-axis current can be expressed as

\[ i_d = -\frac{\lambda_{fd}}{L_d} \left(1 - \frac{\omega_{base}}{\omega_e}\right) \]  (4-41)

\[ i_q = \frac{i_{qb} \omega_{base}}{\omega_e} \]  (4-42)

More details of this control and other controls can be found in [33].
CHAPTER 5
ELECTRIC VEHICLE MODEL

The dynamic model of an EV system is implemented in MATLAB Simulink/SimPowerSystems and SimDriveline environment. Further, the corresponding EV model is constructed out of current commercial components, first, to simulate more realistically and second, to observe the effectiveness of the neural network controller with the off-the-shelf electric motor. Table 5.1 shows the description of the components, whereas Table 5.2 denotes the overall specification of EV. Moreover, the inverter block from Opal-RT is implemented, instead of from SimPowerSystems, because the inverter block from Opal-RT simulates faster than using the inverter block from SimPowerSystems.

The section is divided into four subsections. Each subsection introduces the four individual blocks, the functions of the block, corresponding off-the-shelf component(s) if applicable, and SimPowerSystem or SimDriveline components. Before diving into the discussion, the overall schematic of the EV is shown in Fig. 5.1 while Fig. 5.2 illustrates the overall block diagram of the EV. Lastly, the simulated models in Simulink are shown in Fig. 5.3.
Figure 5.1: Overall schematic of the EV

Figure 5.2: Overall block diagram of the EV

Figure 5.3: Overall block diagram of the EV in Simulink
<table>
<thead>
<tr>
<th>Component</th>
<th>Simulink Block Name</th>
<th>Part Name</th>
</tr>
</thead>
<tbody>
<tr>
<td>Motor</td>
<td>Permanent Magnet Synchronous Machine (SimPowerSystem)</td>
<td>Parker GVM210-150P [34]</td>
</tr>
<tr>
<td>Inverter</td>
<td>IGBT/Diode in Universal Bridge (SimPowerSystem)</td>
<td>Rinehart PM100DX [35]</td>
</tr>
<tr>
<td>ESS</td>
<td>Battery/Resistor/Ideal Switch (SimPowerSystem)</td>
<td>A123 7x15s3p pack [36]</td>
</tr>
<tr>
<td>Differential</td>
<td>Differential (SimDriveline)</td>
<td></td>
</tr>
<tr>
<td>Chassis</td>
<td>Vehicle Body (SimDriveline)</td>
<td>Midsize sedan</td>
</tr>
<tr>
<td>Tire</td>
<td>Tire (Magic Formula) (SimDriveline)</td>
<td>215/55R17 [37]</td>
</tr>
<tr>
<td>Shaft</td>
<td>Flexible Shaft (SimDriveline)</td>
<td></td>
</tr>
<tr>
<td>Brake</td>
<td>Double Shoe Brake (SimDriveline)</td>
<td>Hydraulic</td>
</tr>
</tbody>
</table>

Table 5.1: Part and Simulink block name for components

<table>
<thead>
<tr>
<th>Parameter</th>
<th>Description</th>
<th>Values</th>
</tr>
</thead>
<tbody>
<tr>
<td><strong>Vehicle Dynamic</strong></td>
<td></td>
<td></td>
</tr>
<tr>
<td>$M_{veh}$</td>
<td>Total mass of the vehicle</td>
<td>1,560 kg</td>
</tr>
<tr>
<td>$C_{rr}$</td>
<td>Coefficient of rolling resistance</td>
<td>0.01</td>
</tr>
<tr>
<td>$C_{drag}$</td>
<td>Coefficient of aerodynamic drag</td>
<td>0.35</td>
</tr>
<tr>
<td>$C_{dA}$</td>
<td>Frontal area</td>
<td>0.76 m²</td>
</tr>
<tr>
<td>$\rho$</td>
<td>Air density</td>
<td>1.2 kg/m³</td>
</tr>
<tr>
<td>$R_{wheel}$</td>
<td>Wheel radius</td>
<td>0.324 m</td>
</tr>
<tr>
<td>$\alpha$, $\beta$, A, B, C</td>
<td>See the description in Section 5.</td>
<td></td>
</tr>
<tr>
<td><strong>Tire</strong></td>
<td></td>
<td></td>
</tr>
<tr>
<td>$P_{peak}$ and $P_{cont}$</td>
<td>Peak/ Continuous output power</td>
<td>100/70 kW</td>
</tr>
<tr>
<td>$T_{peak}$ and $T_{cont}$</td>
<td>Peak/ Continuous output torque</td>
<td>256/122 Nm</td>
</tr>
<tr>
<td>$\omega_{max}$ and $\omega_{rated}$</td>
<td>Max/ Rated motor speed</td>
<td>7,500/5480 RPM</td>
</tr>
<tr>
<td>$R_s$</td>
<td>Stator resistance</td>
<td>18.5 mΩ</td>
</tr>
<tr>
<td>$L_d$ and $L_q$</td>
<td>d and q-axis inductance</td>
<td>0.22 and 0.32 mH</td>
</tr>
<tr>
<td>$\lambda_f$</td>
<td>Flux linkage constant</td>
<td>0.0595 Wb</td>
</tr>
<tr>
<td>$P$</td>
<td># of poles</td>
<td>12</td>
</tr>
<tr>
<td>$J_{eq}$</td>
<td>Rotor inertia</td>
<td>0.028 kg·m²</td>
</tr>
<tr>
<td><strong>Battery (7x 15s x 3p)</strong></td>
<td></td>
<td></td>
</tr>
<tr>
<td>$V_{nom}$</td>
<td>Nominal voltage</td>
<td>340 V</td>
</tr>
<tr>
<td>$I_{dis}$  and $I_{chg}$</td>
<td>Nominal discharge and charge current</td>
<td>180/60 A</td>
</tr>
<tr>
<td>$I_{pdis}$ and $I_{pchg}$</td>
<td>Peak discharge and charge current</td>
<td>612/300 A</td>
</tr>
<tr>
<td>Cap</td>
<td>Pack capacity</td>
<td>58.8 Ah</td>
</tr>
<tr>
<td>$E_{nom}$</td>
<td>Pack energy capacity</td>
<td>18.9 kWh</td>
</tr>
<tr>
<td><strong>Shaft and Differential</strong></td>
<td></td>
<td></td>
</tr>
<tr>
<td>FDR</td>
<td>Final drive ratio</td>
<td>8.2</td>
</tr>
<tr>
<td>$S_{mat}$</td>
<td>Shaft material</td>
<td>Al alloy</td>
</tr>
<tr>
<td>$S_{dia}$</td>
<td>Shaft diameter</td>
<td>65 mm</td>
</tr>
<tr>
<td>$S_{length}$</td>
<td>Shaft length</td>
<td>761 mm</td>
</tr>
</tbody>
</table>

Table 5.2: EV specification
5.1 Driver

The simulation starts its operation from Driver block. A block diagram of the driver system is given in Figure 5.4. The primary purpose of this block is producing the applied pedal position (APP), which describes how much the pedal is pushed by the driver. The range of the APP is -100% to 100%. The negative APP denotes braking. When the APP is negative, either hydraulic or regenerative, or both, braking are initiated, depending on the condition.

Inputs to the driver system are reference and actual speed of the vehicle while outputs are APP in percent and reference hydraulic brake force. For this simulation, three different drivecycles are used for the reference speed: UDDS, HWFET, and US06. Fig. 5.5 shows UDDS, HWFET, and US06, respectively. An UDDS drivecycle provides common urban drive trace while a HWFET denotes to highway driving. Lastly, an US06 is for aggressive city driving plus highway. The actual vehicle speed is calculated from chassis and tire block.

Figure 5.4: Block diagram in Driver block
Figure 5.5: a) UDDS, b) HWFET, and c) US06 drivecycle

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First, the driver block receives two speed inputs and converts their units from mph to m/s. Then, these converted inputs are subtracted and applied to a PI controller. The proportional (P) and integral (I) gains that were used are 50 and 1, respectively. The output of the PI controller is APP. Then, this APP is applied to Stateflow block, which is a control logic tool used to emulate the state machine within Simulink model. The main role of this block is determining the required hydraulic brake force from APP. The mask view of the Stateflow diagram is shown in Fig 5.6. Detailed information is covered in subsection 5.4.2.

![Stateflow Diagram](image)

**Figure 5.6: Stateflow diagram of the braking shift logic**

### 5.2 Motor Drive and ESS

The main objective of this block is translating the APP into the actual motor traction torque. Fig. 5.7 illustrates the overall block diagram of the motor drive and energy storage system (ESS) block while Fig. 5.8 shows the overall schematic of the motor drive. As Fig. 5.7 shows, four blocks are identified: max torque, APP2RefTrq, Controller, Motor and Inverter, and ESS.
5.2.1 ‘Max torque’ Block

A block diagram of max torque block is shown in Fig. 5.9. The main purpose of this block is determining the maximum allowable torque. As Fig 5.9 illustrates, two look-up tables are connected with the input vehicle speed, where an upper look-up table contains the continuous torque values while a lower table contains the peak torque values. The temperature is taken into account by constraining the time allowed above continuous torque. A block called “Heat index” is used to accomplish the tasks. The mask view of this block is given in Fig. 5.10. The following equation is utilized to find the shifting point.

\[
\begin{align*}
MTPA & = \frac{APP(\%)}{Motor \ Spd \ (\text{rad/s})} \\
T^* & (Nm) \\
\text{Max Torque} & (Nm) \\
Vdc \ (V) \\
Id^* & \\
Iq^* & \\
\alpha & \\
\beta & \\
\text{IPM} & \\
s & \\
sd & \\
ed & \\
e & \\
abc & \\
dq & \\
Iabc & \\
Id & \\
Iq & \\
\theta & \\
\theta & \\
\text{PWM} & \\
SV & \\
pAFM & \\
\end{align*}
\]
\[ \text{Heat index} = \int \frac{0.3}{t_{max}} \left( \frac{|\text{Torque}_{active}|}{\text{Torque}_{cont}} - 1 \right) \] (5-1)

where \( \text{Torque}_{active} \) is the actual motor torque; \( \text{Torque}_{cont} \) is the continuous motor torque; and \( t_{max} \) is the maximum allowable time for maximum torque to be on. In this model, 10 second is used for \( t_{max} \). The range of the heat index is between 0 and 1. If the heat index is 0, the peak torque is utilized while if it is 1, the continuous torque is used. If the heat index is between two values, the torque linearly decreases from the peak torque to the continuous torque. Simply saying, even if the driver still wants to maintain the full throttle of the motor after \( t_{max} \), this calculation prevents the driver to doing so in order to protect the motor. Furthermore, this block also works properly when the motor operates as a generator.

\[ \begin{array}{c}
\text{speed} \\
\downarrow \\
\text{torque} \\
\downarrow \\
1-0 \text{ Torq} \\
\downarrow \\
\text{Cont torq max} \\
\downarrow \\
\text{torque} \\
\downarrow \\
\text{Heat index} \\
\downarrow \\
\text{Heat index} \\
\downarrow \\
\text{Peak torq max} \\
\downarrow \\
1-0 \text{ Torq} \\
\downarrow \\
\text{Cont torq max} \\
\downarrow \\
\text{Heat index} \\
\downarrow \\
\text{Heat index} \\
\downarrow \\
\text{torque} \\
\downarrow \\
\text{max torque} \\
\end{array} \]

Figure 5.9: ‘Max torque’ block diagram
5.2.2 ‘APP to Reference Torque’ Block

A block diagram of APP2RefTrq block is displayed in Figure 5.11. The role of this block is receiving the maximum torque from max torque block and APP in percent, and converting to the reference torque by multiplying two signals as in Fig. 5.11. Since APP is in percent, it is multiplied by $0.01$ to change decimal value.

5.2.3 ‘Controller’ Block

A block diagram of the controller block is shown in Fig. 5.12. The objective of this block
is translating the reference torque into corresponding PWM signal for the inverter. Inputs to the system are the motor speed, three-phase currents, reference torque, and dc-link voltage. Output is PWM signal.

![Figure 5.12: ‘Controller’ block diagram](image)

After receiving above inputs, the block translates the demanded torque to reference d and q-axis currents by using MTPA control that was discussed in Chapter 4. Simultaneously, the block transforms three-phase currents to d and q-axis and calculates the angular position of the motor through integrating the motor speed. Note that Ferrari’s method is used for MTPA control in this thesis. Then, those signals are applied to the proposed NN controller. The detailed descriptions of the proposed NN controller are explained in Chapter 3. Note that the error is equaled to actual minus reference, not the other way around. Then, the NN controller produces the normalized d- and q-axis voltage. After de-normalizing the voltage by multiplying the space vector PWM (SVPWM) gains, the d- and q-axis voltages are converted to three-phase abc voltages. Lastly, through SVPWM, 6 pulses of PWM signals are calculated.

### 5.2.4 ‘Motor and Inverter’ Block
A block diagram of motor and inverter block is given in Fig. 5.13. Inputs to the system are the shaft speed, 6 SVPWM pulses, and DC-link voltage of ESS. Outputs are three-phase current and motor torque. As Fig. 5.13 illustrates, the inverter from Opal-RT and the motor from SimPowerSystem library are used. Further, notice that the motor takes the speed as its load input, rather than the torque. The primary reason is that the torque is the factor is changing. If the speed was changing, the motor will take the load torque.

The operation of this block is simple. First, the block receives the pulses from the motor and controller block and applies to the inverter for controller. The DC side of the inverter is connected with ESS. With using above inputs, the inverter transforms DC input to AC output accordingly. Then, these AC signals flow through the measurement tool and finally apply to the motor. Lastly, the motor produces the electromagnetic torque.

As for the electric motor component, Parker GVM210-150P was chosen. The motor specification is given in Table 5.2. The primary reason to choose this motor was it possesses 1) reasonable overall peak, 2) continuous power density, 3) operating voltage rating, and 4) low torque ripple for EV application. Fig. 5.14a shows the actual motor while Fig. 5.14b the graph that consists of base speed, peak power, and rated power [34].
For the inverter component, Rinehart PM100DX inverter was used because it not only meets the requirement of the EV system, but also has reasonable size to be fitted in actual EV platform. Fig. 5.15 shows the drawing of PM100DX inverter [35]. As Fig. 5.15 shows, the size of the inverter is only 314 x 200 x 87 mm$^3$.

Figure 5.14: a) Actual motor and b) specification in graph [34]
5.2.5 ‘ESS’ Block

The last block to discuss is an energy storage system (ESS). Generally, five electrical components are formulated to structure the ESS. They are a current sensing modules (CSM), a battery control module (BCM), an electronic distribution module (EDM), and an emergency disconnect switch (EDS). Table 5.3 describes the functionality of each component.

<table>
<thead>
<tr>
<th>Name</th>
<th>Description</th>
</tr>
</thead>
<tbody>
<tr>
<td>CSM</td>
<td>Measures the current through magnetic field. If the current and elapsed time exceed limit value, it sends an alert to BCM. Then, BCM orders EDS to open its switch.</td>
</tr>
<tr>
<td>BCM</td>
<td>Functions as a brain. It disconnects the ESS from the system when the current and voltage exceeds its limit. It communicates with supervisory controller through CAN.</td>
</tr>
<tr>
<td>EDM</td>
<td>Consists of three contactors and one resistor. Three contactors are positive, negative, and pre-charge contactors. Pre-charge function limits inrush current.</td>
</tr>
<tr>
<td>EDS</td>
<td>Disconnects the current flow when BCM alerts. It is usually located between modules. See Figure 5.16.</td>
</tr>
</tbody>
</table>

Table 5.3: Descriptions of ESS components
A block diagram of ESS is displayed in Fig. 5.16. The primary purpose of this block is providing the electrical energy to corresponding motor drives. In lieu of implementing five ESS components, this thesis utilizes four components to emulate the performance of ESS for simplicity. They are batteries, EDM, EDS, and CSM. First, the battery block from SimPowerSystems library is used for battery. The required parameters of the battery are extracted from the datasheet. Second, one $25\ \Omega$ resistor and three ideal switches in SimPowerSystems library are used to model the EDM, where the resistor is representing the pre-resistor while two upper and one lower ideal switches represent the pre-contactor and main positive contactor and negative contactor, respectively. Third, one ideal switch is used to model the EDS. If state of charge (SOC) is less than 20%, the switch will open to prevent the battery to be full discharged. Lastly, the current measurement block in SimPowerSystem library is used to model the CSM.

![Overall block diagram of the ESS]

Figure 5.16: Overall block diagram of the ESS

As for the ESS component, A123 ESS was used. The battery is formulated into 7x15s3p structure, where 7 denotes the number of module, 15s refers to 15 cells in series, and 3p conveys parallel of three. The multiple concatenated cells make a module while the multiple connected
modules construct a pack. Usually, the voltage rating of a cell is on the order of one to six volts. Fig. 5.17a shows the structure of 1 module while Fig. 5.17b shows an actual A123 ESS, which includes 7 modules and other supplementary components.

![Figure 5.17: a) Schematic of 1 module and b) overall ESS [35]](image)

### 5.3 Shaft and Differential

The next block to be discussed is ‘shaft and differential’ block. The primary purpose of the shaft is connecting the motor and differential while the objective of the differential is splitting the shaft torque into two torques for front wheels. Note that only longitudinal motion was considered. Thus, the differential in this thesis will halve the input torque evenly. However, in real environment, the torque is unevenly split when the vehicle is on the curve. Fig. 5.18 shows the overall block diagram of ‘Shaft and Differential’ block.

As Fig. 5.18 shows, two inputs are used. The first input is the motor torque while other is the actual vehicle speed in meters per second. First, to accommodate proper connectivity between Simulink and SimDriveline, Simulink to PS torque block is used. Like the name
implies, the main purpose of this block is translating Simulink signal to mechanical signal. Fig. 5.19 illustrates the Simulink to PS Torque block diagram.

![Overall block diagram of ‘Shaft and Differential’ block](image1)

Figure 5.18: Overall block diagram of ‘Shaft and Differential’ block

![Simulink to PS Torque block diagram](image2)

Figure 5.19: Simulink to PS Torque block diagram

Then, this mechanical torque is now applied to the motor shaft. A block called Flexible shaft in SimDriveline is used to model the actual shaft. As it shows, the total length of the shaft is 1,386 mm. Two shafts were used to prevent the shaft to be shattered from the rotational force, and U-joint is connecting two shafts. The diameter of first shaft that is connected with the motor is 60 mm while the second that is connected with the differential is 65 mm. Other parameters of the shaft can be found in Table 5.2. Then, the output torque of the shaft splits into two torques by
using differential, which is also provided by SimDriveline library. The value 8.2 was used for final drive ratio (FDR).

The second task of this block is translating the actual vehicle speed from the chassis to the motor speed. This is easily achieved by utilizing this equation.

\[
\omega_{motor} = V_{vehicle} \frac{FDR}{R_{tire}}
\]  

(5-2)

where \( R_{tire} \) is the radius of the tire.

### 5.4 Tire, Chassis, and Brake

This subsection contains four tires, four hydraulic brakes, and a vehicle body. The origin of these components is from SimDriveline. The primary purpose of this block is modeling the function of tires, hydraulic brakes, and chassis. Table 5.1 denotes the name of SimDriveline components with associated components in real life while Fig. 5.21 shows the inside view of ‘Tire, chassis, and brake’ block.

As Fig. 5.21 illustrates, inputs to the system are hydraulic brake command from the driver block and two torques from the ‘shaft and differential’ block. Output is the actual vehicle speed in meters per seconds. First, the mechanical signals from the differential and hydraulic brake command are applied to the front left and right wheels. Then, the wheel block produces corresponding signals to vehicle body. These signals are explained in following sections.
Figure 5.20: Overall diagram of ‘Tire, Chassis, and Brake’ block

5.4.1 Tire

A block diagram of wheel block is given in Fig. 5.22. As for the tire, the wheel block in SimDriveline library is used. This block is based on the magic formula, which is invented by Hans Bastiaan Pacejka and called “magic” because only 10 to 20 coefficients are used to represent the vertical load force on the tire, camber angle, and slip angle [38]. The following equation is the general form of the magic formula:

\[ R(k) = d \cdot \sin\{e \cdot \arctan[b(1-e)k + e \cdot \arctan(bk)]\} \]  

(5-3)

in which \(b, c, d,\) and \(e\) are fitting constants while \(R(k)\) is a resultant force from a slip parameter \(k\) [38].

First, the model takes the input torque from the differential in port A and transmits the thrust generated by the tire to the remainder of the vehicle body through port H. Then, the vehicle body calculates the normal force acting on the tire and feeds to port N [39]. Further, pressure and velocity dependent rolling resistance is included for more realistic The explanation
can be found in [39] while the associated coefficients are -0.4736, 1.031, 6.839e-2, 1.65e-4, and 4.125e-7 for \( \alpha \), \( \beta \), A, B, and C, respectively. Finally, since no traction force is transmitted to the rear tire, at port A, a simple inertia block is inserted to represent free rolling.

5.4.2 Brake

One of the distinguished factors of the EV is the regenerative braking, which accumulates the electricity by providing the negative torque to the motor when the vehicle is decelerating. However, having only regenerative braking is not sufficient to meet the total demanded braking force, especially when sudden braking is needed. Further, not only that, the regenerative braking
technique cannot be used when the battery is near to the full charge and is not efficient to use when the vehicle speed is below 10 mph. As a result, the conventional hydraulic brake is still required to compensate the aforementioned problems.

When both braking system is used, there are two possible brake combinations: series and parallel braking. The advantage of the series braking is that the optimal braking combination can be achieved while the advantage of the parallel braking is simple construction. However, the study showed that the parallel braking cannot recover the maximum energy [40].

In this paper, simple series braking is considered, and these are the rules. Only regenerative braking is applied when APP is below -50% while vehicle speed is above 10 mph, and when the vehicle speed is below 10 mph, only hydraulic braking is applied. If the vehicle speed is above 10 mph while APP is above -50%, the motor regenerates only -50% of the total braking force while others are compensated from the hydraulic braking.

A block called Double-Shoe Brake in SimDriveline library is used to emulate the mechanical brake. As Fig. 5.22 shows, it has two connections: Port S and F, where Port S is connected with the input of the tire, which is the output of the differential, and Port F is connected with the demanded hydraulic force from the Driver block. This block takes the demanded brake force and then, translates the physical signal into mechanical signal, so it can be applied in the overall system. This model was chosen because one of the EV examples in Simulink used this as its brake.

### 5.4.3 Chassis

A block called Vehicle body in SimDriveline library is used to model the chassis. The primary objective of this block is to calculate the vehicle longitudinal velocity. This is achieved
by using the total traction force equation.

\[
F = \frac{1}{2} C_{drag} \rho C_{dA} (V_x - V_w)^2 \cdot \text{sgn}(V_x - V_w) + F_{aero} + M_{veh} g \cdot \sin(\phi) + M_{veh} V_x' 
\]

(5-4)

where \( F \) is the total traction force, \( F_{aero} \) is the aerodynamic drag force, \( V_x \) and \( V_w \) is the velocity of the vehicle and wind, respectively, \( F_{rr} \) is the rolling resistance, \( F_{grade} \) is the grade resistance force, \( \phi \) is the angle of the grade, and \( V_x' \) is the acceleration. The vehicle is assumed to be midsize car; thus, the vehicle mass is 1,560 kg, which is average of midsize car [41]. And frontal area and drag coefficient is changed as shown in Table 5.2. Other than these three parameters, default values are used for other parameters. Through this equation, the vehicle speed is calculated.

The next calculation is determining the normal force for the front and rear-axle wheel and can be found by using below equations.

\[
F_{zf} = \frac{-h(F_{aero} + W_{veh} \cdot \sin(\phi) + M_{veh} V_x') + b \cdot W_{veh} \cdot \cos(\phi)}{n(a + b)}
\]

(5-5)

\[
F_{zf} = \frac{h(F_{aero} + W_{veh} \cdot \sin(\phi) + M_{veh} V_x') + a \cdot W_{veh} \cdot \cos(\phi)}{n(a + b)}
\]

(5-6)

in which \( h \) is the height of vehicle, which is center gravity above the ground, \( a \) and \( b \) is distance of front and rear axles from the normal projection point of vehicle CG, and \( W_{veh} \) is the weight of the vehicle, which is mass of the vehicle times the gravity. These normal forces are applied to the wheels.
CHAPTER 6
SIMULATION RESULTS

For overall simulation, MATLAB Simulink is used because it has 1) easy graphical user interface (GUI), 2) simple data handling and storage, 3) extensive component libraries, such as mechanical, electrical, vehicle, and others, and 4) the flexibility in transition between physical and other signals. Three different simulations were conducted to observe the effectiveness of the proposed ANN controller. In the first, a basic feedback speed-loop control for SPMSM was observed when the motor 1) is accelerating and decelerating, 2) experiences parameter variation 3) experiences load torque variation, and 4) is in high speed. In the second, a basic feedback speed-loop control for IPMSM was tested to observe the performance in same conditions as in the first simulation. In addition, for both simulations, the conventional PI control was also observed to compare with the proposed control. In the last simulation, the NN control for IPMSM was implemented in electric vehicle model as illustrated in Chapter 5.

6.1 SPMSM Simulation Result

6.1.1 Motor Parameter and Simulation Setup

First, 0.2 kW SPMSM motor was used to validate the controls. The specifications of 0.2 kW motor are given in Table 6.1.

Conventionally, most motor drives have widely employed analog-based controller. However, with the advent of the digital signal processor (DSP) and microcontroller (μC), nearly
all of motor drives employs digital controller to control the motor because the digital controller offers 1) more powerful performance and 2) flexibility than the analog-based controller. Further, generally, the digital controller samples the analog signal at certain controller. Further, generally, the analog signal at certain rate. Throughout this thesis, 0.1 ms sampling rate was used because it is not only reflecting typical sampling rate for motor drive, but also not causing any “data freeze”, especially in hardware experiment.

<table>
<thead>
<tr>
<th>Parameter</th>
<th>Value</th>
<th>Units</th>
</tr>
</thead>
<tbody>
<tr>
<td>Rated Power</td>
<td>200</td>
<td>W</td>
</tr>
<tr>
<td>dc voltage</td>
<td>42</td>
<td>V</td>
</tr>
<tr>
<td>Permanent magnet flux</td>
<td>0.0154</td>
<td>Wb</td>
</tr>
<tr>
<td>Inductance in q-axis, $L_q$</td>
<td>0.255</td>
<td>mH</td>
</tr>
<tr>
<td>Inductance in d-axis, $L_d$</td>
<td>0.255</td>
<td>mH</td>
</tr>
<tr>
<td>Stator copper resistance, $R_s$</td>
<td>0.22</td>
<td>Ω</td>
</tr>
<tr>
<td>Inertia</td>
<td>0.07</td>
<td>kg·m²</td>
</tr>
<tr>
<td>Damping coefficient</td>
<td>0.01</td>
<td></td>
</tr>
<tr>
<td>Pole pairs</td>
<td>4</td>
<td></td>
</tr>
</tbody>
</table>

Table 6.1: 0.2 kW PMSM parameter in simulation and hardware study

Unlike the digital controller, the inverter and motor do not have any sampling rate; thus, these components need another solution. For the solution, 2 μs sampling rate was used to model the continuous state as close as possible. Lower sampling rate is better, but it increases the simulation time, simultaneously.

As shown in Fig. 3.1 and 3.10, both controllers are using PI controller for their speed-loop controller. To tune PI gains, first, a cross-over frequency of the speed loop was chosen. Upon speculations, it was concluded that 2.5 kHz provided the most adequate response. Then, phase margin was chosen. Upon speculations, 60 degree was chosen for same reason. Then, the following equations were used to determine the initial PI gains of the speed loop controller.

$$\omega_{cross\_freq} = 2500$$

(6-1)
where \(J_{eq}\) is the inertia of the motor. With having these initial PI gains as base values, slight modification was done. As a result, the final \(K_p\) and \(K_i\) were 3.83 and 79.12, respectively.

Likewise, the PI gains of the current controller for the conventional controller were calculated. Because the inner current loop controller has to respond faster than the outer speed loop controller, the crossover frequency of the current controller had to be a magnitude larger than the crossover frequency of the speed controller. For phase margin, same value as in speed-loop controller was used. The following calculations were used to determine initial PI gains.

\[
\omega_{\text{crossfreq}} = 25000
\]

\[
PM = 60 \times \frac{\pi}{180}
\]

\[
W_{ci_{kpki}} = \tan (PM - \frac{\pi}{2} + \tan^{-1}\left(\omega_{\text{crossfreq}} \times \frac{L_d}{R_s}\right))
\]

\[
K_{ii} = \omega_{\text{crossfreq}} \times \sqrt{\frac{L_d^2 + \left(\omega_{\text{crossfreq}} \times L_d\right)^2}{W_{ci_{kpki}}^2 + 1}}
\]

\[
K_{pp} = K_{ii} \times \frac{W_{ci_{kpki}}}{\omega_{\text{crossfreq}}}
\]

Similarly, the calculated PI gains were modified. As a result, for the PI gains, 1.38 and 199.8 were used, respectively.
Aforementioned, total of four tests were conducted to observe the performance of the proposed controller and conventional controller. They are acceleration and deceleration test, load torque variation test, parameter variation test, and flux weakening test. Note that the initial load torque of 0.1 Nm was used.

First, the acceleration and deceleration test was conducted as follows. At $t=0.56$ s, the input speed was increased from 0 rad/sec to 100 rad/sec, and then stayed at 100 rad/sec for 1 second. At $t=1.56$s, the speed was decreased to 50 rad/sec and retained its value for 1 second. At $t=2.56$ s, the speed was increased to 200 rad/sec and decreased again to 100 rad/sec at $t=3.56$ s.

After the acceleration and deceleration test, the load torque was varied as follows. At $t=4.36$ s, the load torque was changed from 0.1 to 0.3 Nm. At $t=5$ s, the load torque was dropped to 0.1 Nm.

After the load torque variation test, the parameter variation test was conducted as follows. At $t=5.5$ s, the stator resistance, synchronous inductance, and flux linkage constant was changed: the stator resistance was doubled while the inductance and flux linkage constant was reduced 20%.

Lastly, the flux weakening control was tested for both controllers. At $t=6.75$ s, the reference speed was accelerated at a rate of 150 rad/s² until it reached just slightly below the maximum speed, which is about 400 rad/s. At $t=9.75$ s, the reference speed was decreased to 200 rad/s to check whether the control can return to normal control from the flux weakening control.

6.1.2 Results

The traction result of speed, dq currents, abc currents, and normalized system voltage for both conventional and neural network controller are shown in following figures (from Figure 6.1
to 6.7), respectively.

As the figures show, both controllers perform similarly until the flux weakening test. As Fig. 6.1 shows, when the reference speed is about 250 rad/s, the conventional control loses the stability. Even though the PI gains for conventional control were changed in numerous times, none of the PI gains could solve the problem. However, unlike the conventional control, ANN control exhibits very stable speed traction in high speed region. When the motor reaches 400 rad/sec, as Fig. 6.2 shows, the d-axis current is producing the negative value to prevent the back-EMF to exceed the maximum voltage (Fig 6.6).

Figure 6.1: Speed traction of SPMSM (simulation)
Figure 6.2: ANN d- and q-axis current traction of SPMSM (simulation)

Figure 6.3: Conventional d- and q-axis current traction of SPMSM (simulation)
Figure 6.4: ANN abc current traction of SPMSM (simulation)

Figure 6.5: Conventional abc current traction of SPMSM (simulation)
Figure 6.6: ANN normalized amplitude of voltage of SPMSM (simulation)

Figure 6.7: Conventional normalized amplitude of voltage of SPMSM (simulation)
6.2 IPMSM Simulation Result

6.2.1 Motor Parameter and Setup

100 kW Parker GVM210-150P IPMSM was used to demonstrate the simulation. Table 5.1 shows the specification of the preceding motor. The PI gains for the speed controller for both controllers were calculated using the method that was shown above. Likewise, the PI gains of the current controller were determined.

Same four tests were conducted as follows. First, the acceleration and deceleration test were conducted as follows. At t=0.5 s, the input speed was increased from 0 rad/sec to 100 rad/sec, and then kept at 100 rad/sec for 0.75 second. At t=1.25s, the speed was decreased to 50 rad/sec. At t=2 s, the speed was increased to 200 rad/sec and decreased again to 100 rad/sec at t=3 s. Then, the load torque variation was observed.

Then, the load torque variation test was observed. At t=3.5 s, the load torque was changed from 20 Nm to 100 Nm. At t=4.25 s, the load torque was further increased to 200 Nm and then, dropped to 50 Nm at t=5 s.

After the load torque variation test, the parameter variation test was conducted. At t= 5.5 s, the stator resistance was doubled while the d- and q-axis inductance was decreased 20% and 30%, respectively. Lastly, the flux linkage constant was reduced by 20%. At 6.5s, the parameter was changed back to the original values.

Lastly, at t= 7 s, the flux weakening control was tested by increasing the reference speed at a rate of 200 rad/sec² until 650 rad/s is reached. Then, the speed was decreased to 300 rad/s at t= 10.5 s.
6.2.2 Results

Fig 6.8 and 6.9 shows the speed traction and dq current traction of both controllers while Fig. 6.10 and 6.11 illustrates the abc current traction of ANN control and conventional control for IPMSM respectively. Lastly, Fig. 6.12 shows the normalized system voltage of both controllers.

For the first and second tests, both controllers showed similar performance as the figures illustrate. However, for parameter variation test, the conventional controller shows better response than the proposed controller. As shown in Fig 6.9, the d- and q-axis currents of the ANN controller are oscillating, but not for the conventional controller. Further, in the flux weakening control test, Fig. 6.9 shows that the ANN controller uses less d-axis current than the conventional controller while for q-axis current, the ANN controller uses slightly more current than the conventional controller. Aforementioned, the d-axis current produces the flux against the magnet flux to prevent the back-EMF exceeding certain threshold. Larger the d-axis current, the more the opposing flux is. However, the downside is that if this opposing flux is too large, it can demagnetize the magnet, where the demagnetization of the magnet causes the motor to be not operable. Thus, having less d-axis current is better. In Fig 6.9, the ANN controller produces less d-axis current than the conventional controller; thereby, it can be concluded that the former controller performs better in high speed than the latter controller.
Figure 6.8: Speed traction of IPMSM (simulation)

Figure 6.9: D-and q-axis current traction of IPMSM (simulation)
Figure 6.10: ANN abc current traction of IPMSM (simulation)

Figure 6.11: Conventional abc current traction of IPMSM (simulation)
6.3 Electric Vehicle

To observe the transient response of the proposed NN controller and performance of the regenerative and hydraulic braking in EV application, US06-City drive cycle is used because this drive cycle not only consists of frequent start and stop situation, but also requires the vehicle to go up to 71 mph. In other words, it was chosen because it represents a very practical and typical driving condition in the United States. Further, the proposed controller was compared with PI-based controller, which is widely used in industry, to demonstrate the effectiveness of the controller.

Fig. 6.13 shows the reference and actual vehicle speed while Fig. 6.14 illustrates the reference and actual motor torque. The APP and hydraulic brake force, a dotted line, can be observed in Fig. 6.15. Fig. 6.16 shows the behavior of the d- and q-axis currents. Lastly, Fig.
6.17 exhibits the SOC of the battery and APP graph together to show the functionality of the regenerative breaking.

As Figs. 6.13, 6.14, and 6.15 show, the EV using the proposed NN controller tracks the reference signal closely and shows similar performance with the EV using PI controller. Further, Fig. 6.16 validates that hydraulic brake assists the regenerative braking when the above condition is not met. In addition, when the braking occurs, meaning APP is negative value, the battery is charged (Fig. 6.17).

![Figure 6.13: Speed traction of EV](image-url)
Figure 6.14: Torque traction of EV

Figure 6.15: D- and q-axis current traction of EV
Figure 6.16: ANN APP and hydraulic braking traction of EV

Figure 6.17: ANN APP and SOC traction of EV
CHAPTER 7
HARDWARE RESULTS

7.1 Hardware Experiment Component Description

Five components were used to conduct hardware experiment. They are 1) dSPACE CP/DS1103, 2) Lab-volt inverter, 3) Lab-volt power supply, 4) OP8660 data acquisition interface, 5) motor dyno kit. Figure 7.1 shows the schematic of the overall hardware experiment while Figure 7.2 shows the physical hardware setup of the system.

Figure 7.1: Schematic of the hardware experiment
Figure 7.2: Physical wiring of hardware experiment
For a digital signal processor (DSP) controller, dSPACE CP/DS1103 board was used. As Fig. 7.3 shows, the board contains multiple on-board peripherals such as analog-to-digital (ADC) converters, digital-to-analog (DAC) converter, encoder, CAN and serial interface, and PWM ports. Further, through Real-Time Interface (RTI) and Real-Time Workshop (RTW), Simulink models can be translated to the code that CP1103 board can understand. Inputs can be inserted through either Simulink or software called ControlDesk. Outputs of analog signal can be seen in ControlDesk only, through the I/O board in CP/DS1103 board.

Two inverters were considered for the hardware test. They are MOSFET-based inverter from Vishay-Hirel systems LLC and IGBT-based inverter from Lab-volt. First, the voltage rating of both inverters is 400V while the current rating is 10A. The advantage of the first inverter is that it not only provides two inverter modules in one board, but also allows the direct connections with dSPACE CP/DS1103 controller. The advantage of the second inverter is that it is IGBT-based, which is widely used in current industries, and provides safer and more convenient connections than the first inverter. Further, when the current exceeds 10A, both
controllers automatically cuts the power off to the board and alerts the user by lighting the red LED. Among these two inverters, in this thesis, the latter inverter was used because 1) only one inverter set was enough to conduct the experiment, and 2) the connection between the inverter and other systems only requires banana plug.

Figure 7.4: Vishay-Hirel MOSFET inverter

Figure 7.5: Lab-volt IGBT inverter
For a power supply, Lab-volt 8821 power supply was used. This power supply can provide both constant and variable single phase and three-phase AC and DC power simultaneously. Further, it also provides a 24 V AC output to other Lab-volt system. In this thesis, only variable DC voltage was used to provide the power to DC side of the inverter. The voltage and current ratings are 120V and 8A, respectively.

Figure 7.6: Lab-volt power supply
The next component to discuss is Opal-RT OP8660 data acquisition interface. Figure 7.7 represents the OP8660 data acquisition interface while Table 7.1 shows the pinout descriptions. This component was used to measure the current and voltage associated within the motor drive. Banana jacks are used to measure the current and voltage.

![Figure 7.7: Opal-RT OP8660 data acquisition interface [42]](image)

<table>
<thead>
<tr>
<th>Letter</th>
<th>Connector type</th>
<th>Description</th>
</tr>
</thead>
<tbody>
<tr>
<td>A</td>
<td>DB9</td>
<td>6 pulse inverter output</td>
</tr>
<tr>
<td>B</td>
<td>DB9</td>
<td>Encoder input connector</td>
</tr>
<tr>
<td>C</td>
<td>Banana jacks</td>
<td>Analog/Digital inputs and outputs</td>
</tr>
<tr>
<td>D</td>
<td>Banana jacks</td>
<td>High current probe connectors (Maximum 15A AC)</td>
</tr>
<tr>
<td>E</td>
<td>Banana jacks</td>
<td>High voltage probe connectors (Maximum 600V)</td>
</tr>
</tbody>
</table>

Table 7.1: Opal-RT OP8660 data acquisition interface pinouts

The last component is the motor dyno kit. This kit is from Motorsolver. Figure 7.8 shows the overall PMSM motor and DC motor/generator kit while Table 6.1 illustrates the motor and generator parameter. In Fig. 7.8, the right motor is BLDC7105 while the left motor is DCGEN7105. To operate BLDC7105 motor, Phase A, B, and C need to be wired to yellow, green, and blue, respectively. Lastly, the encoder is connected with dSPACE CP1103 board.

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7.2 Software Setup

To communicate between dSPACE I/O board and Simulink, blocks from RTI1103 Board library has to be used. This RTI library is designed to specify the hardware setup for real-time applications. The library is divided into Master PPC (Power PC) and Slave DSP library. In Master PPC library, ADC and encoder blocks are present while in Slave DSP library, PWM blocks are located. Figure 7.9 illustrates the overall block diagram of both controllers. As it exhibits, three main components were used. They are DS1103SL_DSP_PWMSV, DS1103ADC, and DS1103 ENC blocks. The primary role of the first component is providing the SVPWM signals to the inverter while the objective of the second and third component is receiving three-phase currents and position of the motor encoder, respectively. After constructing the schematic
as shown in Fig. 7.9, by using Simulink Coder, formerly Real-Time Workshop, the model code, .sdf file, was generated. This file provides access to the variables, such as reference speed, dq currents, angular displacement, and others, in the model to separate software called ControlDesk. ControlDesk can control parameters and variables in the model in real time, observe the system during tests, and capture the output. An overview of ControlDesk is shown in Figure 7.10.

Figure 7.9: Schematic of SVPWM based hardware experiment
This section illustrates how SVPWM technique is implemented in MATLAB Simulink for dSPACE platform. Figure 7.11 shows the block and inside view of DS1103SL_DSP_PWMS V block. As Fig. 7.14a shows, the block requires three factors: $T_1/T_p$, $T_2/T_p$, and Sector. These variables were determined using following procedures.

First, the plane of reference vectors from the controller block need to be in $\alpha\beta$-plane. If the plane of the reference plane is in dq-plane, below equation can be used to convert.

$$
\begin{bmatrix}
  f_a \\
  f_\beta
\end{bmatrix} =
\begin{bmatrix}
  \cos(\theta) & -\sin(\theta) \\
  \sin(\theta) & \cos(\theta)
\end{bmatrix}
\begin{bmatrix}
  f_a \\
  f_\beta
\end{bmatrix}
$$

(7-1)

where $\theta$ is the rotor position angle in radians. If it is in abc-plane, one can use
\[
\begin{bmatrix}
  f_a \\
  f_b
\end{bmatrix} = \frac{2}{3} \begin{bmatrix}
  1 & -1/2 & -1/2 \\
  0 & -\sqrt{3}/2 & -\sqrt{3}/2
\end{bmatrix} \times \begin{bmatrix}
  f_a \\
  f_b
\end{bmatrix}
\]

(7-2)

Figure 7.11: SVPWM block and description for hardware experiment
Second, the reference vector needs to be written in complex and magnitude and angle forms.

\[ f_\alpha + if_\beta = |f_\alpha|e^{\theta} \]  
(7-3)

\[ |f_\alpha| = \sqrt{f_\alpha^2 + f_\beta^2} \]  
(7-4)

\[ \theta = \tan^{-1}\frac{f_\beta}{f_\alpha} \]  
(7-5)

The instantaneous stator voltage space vector can be written in terms of the inverter output voltages [43].

\[ v_s^a(t) = v_{aN}e^{j\phi} + v_{bN}e^{\frac{j2\pi}{3}} + v_{cN}e^{\frac{j4\pi}{3}} \]  
(7-6)

If the switching functions are implemented in above equation, the equation can be rewritten as

\[ v_s^a(t) = \frac{2}{3}V_d \left( q_a e^{j\phi} + q_b e^{\frac{j2\pi}{3}} + q_c e^{\frac{j4\pi}{3}} \right) \]  
(7-7)

where \( q_a, q_b, \) and \( q_c \) are representing the switching state of the upper switch of the inverter and \( V_d \) is the amplitude of the DC-link voltage. If \( V_d \) is assumed constant, 8 possible combinations of the switching states are observed. They are

\[ v_s^a(000) = v_0 = 0 \]

\[ v_s^a(001) = v_1 = \frac{2}{3}V_d e^{j\phi} \]

\[ v_s^a(010) = v_2 = \frac{2}{3}V_d e^{\frac{j2\pi}{3}} \]  
(7-8)

\[ v_s^a(011) = v_3 = \frac{2}{3}V_d e^{\frac{j\pi}{3}} \]

\[ v_s^a(100) = v_4 = \frac{2}{3}V_d e^{\frac{j4\pi}{3}} \]

\[ v_s^a(101) = v_5 = \frac{2}{3}V_d e^{\frac{j5\pi}{3}} \]

\[ v_s^a(110) = v_6 = \frac{2}{3}V_d e^{j\pi} \]
As it shows, only six of them are active switching state while 000 and 111 are inactive switching state, where inactive means no current is flowing to the motor. Figure 7.12 shows 8 switching states in $\alpha\beta$ graphical plane.

Figure 7.12: Basic voltage vectors [43]

The next step is finding switching time duration in each sector. The basis of the time duration equation is

$$T_1V_{s1} + T_2V_{s2} + T_0V_0 = T_sV_{ref}$$  \hspace{1cm} (7-9)$$

where $V_{ref}$ is the reference voltage; $V_{s1}$ and $V_{s2}$ are the amplitude of each sector; and $T_s$ is the total cycle time, which equals to

$$T_s = T_{s1} + T_{s2} + T_0$$  \hspace{1cm} (7-10)$$
Since the instantaneous voltages are in space vector, the reference voltage needs to be rewritten in terms of space vector

\[ V_{ref} = V_{ref}e^{j\theta} \quad (7-11) \]

To clarify the concept more, the calculation of the time duration of Sector 1 is shown in following paragraphs. If the sector is in 1, the amplitude of the sector is equalled to

\[ V_{s1} = \frac{2V_d}{3}, V_{s2} = \frac{2}{3}V_d e^{j\frac{2\pi}{3}}, V_0 = 0 \quad (7-12) \]

If Eq. 7-11 and 7-12 are inserted into Eq. 7-9, the equation can be written as

\[ T_s V_{ref} \cos(\theta) = T_1 \frac{2}{3}V_d + T_2 \frac{1}{3}V_d \quad (7-13) \]

\[ T_s V_{ref} \sin(\theta) = T_2 \frac{\sqrt{3}}{3}V_d \quad (7-14) \]

\( T_1 \) and \( T_2 \) for Sector 1 is given by

\[ T_1 = T_s \frac{\sqrt{3}V_{ref}}{V_d} \sin \left( \frac{\pi}{3} - \theta \right) \quad (7-15) \]

\[ T_2 = T_s \frac{\sqrt{3}V_{ref}}{V_d} \sin(\theta) \quad (7-16) \]

Figure 7.13 shows a space vector diagram for sector 1.

Figure 7.13: Space vector diagram for sector 1, described with a) the duty cycle, b) switching states [43]
For remaining sectors, the time duration can be calculated by

\[
T_1 = T_c \frac{\sqrt{3} V_{\text{ref}}}{v_d} \sin \left( \frac{\pi}{3} - \theta + \frac{n-1}{3} \pi \right) \quad (7-17)
\]

\[
T_2 = T_c \frac{\sqrt{3} V_{\text{ref}}}{v_d} \sin \left( \theta - \frac{n-1}{3} \pi \right) \quad (7-18)
\]

\[
T_0 = T_s - T_1 - T_2 \quad (7-19)
\]

where \( n \) is the sector number. Since the time durations for each sector are determined, the next procedure is to arrange the duty cycles for each legs of the inverter. To accomplish the task, the switching time is divided amongst the 7 switching states and expresses as

\[
T_s = \frac{T_0}{4} + \frac{T_1}{2} + \frac{T_2}{2} + \frac{T_0}{2} + \frac{T_2}{2} + \frac{T_1}{2} + \frac{T_0}{4} \quad (7-20)
\]

Table 7.2 summarizes \( T_1, T_2 \) and \( T_0 \) for each sector while Figure 7.14 shows the resultant duration times for each sector. Finally, Figure 7.15 shows the mask block diagram view of calculating \( T_1/T_p \) and \( T_2/T_p \) in Simulink.

<table>
<thead>
<tr>
<th>Sector</th>
<th>Duration times</th>
</tr>
</thead>
<tbody>
<tr>
<td></td>
<td>( T_1 )</td>
</tr>
<tr>
<td>1</td>
<td>( T_c \cdot a \cdot \sin(\pi/3 - \theta) )</td>
</tr>
<tr>
<td>2</td>
<td>( T_c \cdot a \cdot \sin(2\pi/3 - \theta) )</td>
</tr>
<tr>
<td>3</td>
<td>( T_c \cdot a \cdot \sin(\pi - \theta) )</td>
</tr>
<tr>
<td>4</td>
<td>( T_c \cdot a \cdot \sin(4\pi/3 - \theta) )</td>
</tr>
<tr>
<td>5</td>
<td>( T_c \cdot a \cdot \sin(5\pi/3 - \theta) )</td>
</tr>
<tr>
<td>6</td>
<td>( T_c \cdot a \cdot \sin(2\pi - \theta) )</td>
</tr>
</tbody>
</table>

Table 7.2: Duration time for each sector
Figure 7.14: Waveform showing sequence of switching states for each sector [43]
7.4 Hardware Setup

Before examining the hardware setup, note that the steps do not have to be followed these steps, but could start differently. The following procedures were used to conduct the hardware test.

1. Connect the banana plug cables to the positive (7) and negative (N) of variable DC power supply, (Labvolt 8821-20) with the positive and negative terminals of the digital multimeter. If the designated power supply is not present, using alternative power supply that meets the voltage and current rating can be used.

2. Connect the positive (7) and negative terminal (N) of the power supply to the positive (1 or +HV) and negative terminal (2 or GND) of the inverter. Then, connect the positive and negative terminal of the inverter with the positive and negative terminal of E9.

3. Then, connect the three current outputs (3, 4, and 5 of IGBT or Phase A1, B1, and C1 of MOSFET) of inverter with the positive terminals (I9, I10, and I11) of the current
measurement of the data acquisition (OP8660). The reason I9, I10, and I11 terminals were used is the connection on back side of the data acquisition with the analog output [39]. In other words, I1, I2, and I3 or others can be used if the connection on the back side is connected as instructed in [39].

4. Link the negative terminals of I9, I10, and I11 with the ABC terminals of the motor from Figure 7.8.

5. By using BNC cable, connect ADCH17 through 20 with the terminal 1, 2, 3, and 5 of OP8660 Analog Outputs, where ADCH17 through 19 are for three-phase currents while ADCH20 is for DC-link voltage.

6. Connect DB9 cable between Lab-volt IGBT inverter board and Slave I/O port of dSPACE board to send PWM signals.

7. Connect between INC1 and encoder terminal of the motor to measure the rotor position.

8. Adjust the shaft of the PM motor to d-axis. This can be adjusted by using simple small DC power supply. To accomplish the task, connect the positive terminal of power supply to A-phase while negative terminal to B-phase. Increase the voltage little bit. Be cautious while you are increasing the voltage.

7.5 Results

Three tests were conducted to observe the effect: acceleration and deceleration test, load torque test, and flux weakening control test. Since the motor parameter varies automatically as the motor operates, separate parameter variation test wasn’t conducted. Unlike in the simulation, in order to observe the impact of the controller, each test was demonstrated individually. 15 kHz
with 0.1 μs dead-time and 0.1 ms were used for the switching frequency and sampling rate, respectively.

7.5.1 Acceleration and Deceleration Test

At t=0.76 s, the input speed was increased from 0 rad/sec to 100 rad/sec, and then stayed at 100 rad/sec for a second. At t=1.76s, the speed was decreased to 50 rad/sec and kept its value for a second. At t=2.76 s, the speed was increased to 200 rad/sec and decreased again to 100 rad/sec at t=3.76 s.

Figure 7.16 and 7.17 show the speed and dq current traction of both proposed and conventional controller. As it illustrates, both controller performed similarly. On the other hand, a significant difference was observed in the voltage magnitude graph as illustrated in Fig. 7.18. When the speed was increased from 50 rad/s to 200 rad/s, the conventional controller was produced the peak voltage magnitude of 1.12 while the ANN controller produced 1.02. This means that the ANN controller is more efficient than the conventional controller when the motor is accelerating, since the voltage is proportional to current.
Figure 7.16: Speed traction of SPMSM (hardware)

Figure 7.17: Dq-axis current traction of SPMSM (hardware)
7.5.2 Load Torque Variation Test

Load torque was applied by connecting 56.14 Ω resistor across the terminal of DCGEN7105 motor/generator while the PM motor was rotating 200 rad/s. The purpose of this resistor is to form close-loop circuit in the stator, so the current can be flow through the stator of DCGEN7105 motor. The concept of Faraday’s law of induction is applied to accomplish the task. For example, the rotating rotor while no current flows through the stator causes the back-EMF to be induced across the stator of the generator. By Ohm’s law, if a resistor is connected between terminals, a close-loop circuit is created. Then, this current will generate the opposing magnetic field against the back-EMF. Thus, certain load torque will be created. The resistance of 56.14 Ω was selected because it was the lowest resistance that could be found in the lab. As a result, when this resistance was applied, 0.32 A of current was flowed through the stator winding.
Other solutions, such as controlling DCGEN7105 by applying the DC voltage across the terminal and applying the friction manually, were considered and tried to apply it, but none of mentioned solutions worked. The first approach couldn’t be implemented because the current flows through the power supply coil and possibly burn the power supply. The second approach was neglected because same amplitude of the friction and time could not be applied. Therefore, connecting the resistor across the terminals was the only option to apply the load torque.

When the load torque was applied, only minimal impact was observed in both dq current and speed. Thus, it is now shown in this thesis. Unlike them, the significant result was observed in voltage magnitude graph. The result showed that the proposed controller is utilizing less voltage than the conventional controller. As Fig. 7.19 shows, the voltage magnitude of the proposed controller was 0.05 in average less than the conventional controller.

![Figure 7.19: Load torque Vm traction of SPMSM](image)

Figure 7.19: Load torque Vm traction of SPMSM
7.5.3 Flux Weakening Control Test

At \( t=0.5 \) s, the input speed was increased from 0 rad/s to 200 rad/s with slope of 200 rad/s\(^2\). As soon as the speed reaches 200 rad/s, the slope was decreased to 10 rad/s\(^2\) until it reaches 280 rad/s. Then, the input speed was kept for 4 seconds and decreased to 100 rad/s.

Fig. 7.23 represents the speed traction of both controllers. It clearly shows that the conventional controller cannot track the speed reference while the proposed controller can. Further, Fig 7.21 shows that while the proposed controller used the flux weakening method by preventing the voltage magnitude to exceed 1.2, which is maximum voltage magnitude that the motor can achieve, the conventional controller could not. Fig. 7.22 shows the reference d-axis current of both controllers while Fig 7.23 shows the actual dq current tractions. Due to cable and other noise, the actual dq currents were hard to comprehend. However, the reference d-axis that is shown in Fig 7.23 shows that the flux weakening control works well in the proposed controller while not in the conventional controller.

![Figure 7.20: a) Conventional and b) ANN speed traction of SPMSM](image-url)
Figure 7.21: a) Conventional and b) ANN voltage magnitude of SPMSM

Figure 7.22: a) Conventional and b) ANN d-axis reference current of SPMSM

Figure 7.23: a) Conventional and b) ANN actual dq current traction of SPMSM
CHAPTER 8
CONCLUSION

In this thesis, the artificial neural network based vector control for PMSMs has been evaluated by theoretical analysis, simulation, hardware experiment, and comparison with conventional PI-based vector control.

First, the background of the PMSM and other control schemes from other literature were discussed. Through the background search, the PMSM was widely used in applications that need precise and intricate operation and possessed many advantages like high torque density, high efficiency, and others compared to other AC machines. Four other control schemes, such as $H_\infty$, fuzzy logic, matrix inverter fed, and wavelet neural network, were observed. Except few control schemes, most literatures validated their control in only simulation, not hardware experiment. This inspired the author to investigate the control scheme that can function in both simulation and real world.

Second, this thesis included the detailed explanations of the motor parameters that are required by the control. Those parameters are stator resistance, inductances, flux linkage constant, inertia, and more. The investigation covered not only the backgrounds of each parameter, but also how to measure the parameters.

In following section, the conventional PI-based field oriented and novel neural network vector controllers were investigated. In each subsection, both the mathematical representation
and Simulink model were presented. In addition, in neural network section, detailed training mechanisms with examples were covered.

Chapter 4 summarized the MTPA and flux weakening control. The fundamentals and various control schemes of MTPA and flux weakening were explained in detail.

Then, in Chapter 5, this thesis discussed the electric model that was constructed by the author. The electric model was comprised of four major blocks: Driver, Motor drive, Shaft and Differential, and Vehicle dynamics with chassis, tire, and mechanical brake.

Then, the motor and controllers were simulated using MATLAB/Simulink. To observe the effectiveness of the proposed control, it was compared with the conventional PI-based control. The power rating of SPMSM and IPMSM was 0.2 kW and 100 kW, respectively. 0.2 kW SPMSM was chosen to compare the result of the hardware motor while 100 kW IPMSM was selected to validate the effectiveness of the proposed control in electric vehicle application. Both SPMSM and IPMSM were observed when the motor is 1) accelerating and decelerating, 2) experiencing varying load torque, 3) experiencing varying parameter, and 4) performing the flux weakening control. First, for SPMSM, the proposed control sustained its operation very stably and reliably in the flux weakening region, but the conventional PI-based control could not keep the operation. Second, for IPMSM, the result showed that the NN control utilizes less d-axis current than the conventional controller. This conveys that the NN control has less susceptible for the magnet to be demagnetized. Lastly, the proposed control was demonstrated in electric vehicle model. The result showed that the proposed controller can be a potential replacement of the existing control scheme, such as PID, fuzzy logic, or etc, and provides adequate traction control in EV application.
Finally, this thesis conducted the hardware experiment to verify the proposed control in real environment. 200 W BLDC motor from Motorsolver and dSPACE CP1103 controller were utilized. For clear clarification, this thesis included individual instructions how the hardware experiment was setup. As similar with the simulation, the proposed control was compared with the conventional controller. Three tests were considered and are acceleration and deceleration test, load torque variation test, and flux weakening control test. Like in the simulation, the proposed control performed better in flux weakening region than the conventional control. Even though the proposed control could perform the flux weakening control well, the PI-based controller could not. Further, there was 3.5% increase improvement in maximum speed when the proposed control is used.

8.1 Limitation and Future Work

The major limitation of this work was ignoring the non-linearity and saturation of magnetic materials, including stator and rotor saturation. If these are accounted, the complex control is required. Another limitation is lacking the validation result of IPMSM in hardware. Because the lab did not have any IPMSM motor, this couldn’t be conducted. Thus, the future goal is conducting the hardware test for IPMSM and validating the effectiveness of the proposed control.
REFERENCES


