DESIGN OF PORTABLE ROTARY DRILL SET UP AND DEVELOPMENT OF MATHEMATICAL MODEL FOR ESTIMATING PHYSICO-MECHANICAL PROPERTIES OF IGNEOUS ROCKS THROUGH NOISE MEASUREMENT

Thesis

Submitted in partial fulfilment of the requirements for the degree of

DOCTOR OF PHILOSOPHY

by

MASOOD

RESEARCH SUPERVISORS

Dr. Harsha Vardhan

&

Dr. M. Aruna



DEPARTMENT OF MINING ENGINEERING NATIONAL INSTITUTE OF TECHNOLOGY KARNATAKA, SURATHKAL, MANGALORE – 575025 AUGUST, 2015

DECLARATION

by the

Ph. D. RESEARCH SCHOLAR

I hereby declare that the Research Thesis entitled "Development of Mathematical Models for Determining Physical Properties of Igneous Rocks Through Noise Measurement using Fabricated Portable Drill" which is being submitted to the National Institute of Technology Karnataka, Surathkal in partial fulfillment of the requirements for the award of the Degree of Doctor of Philosophy in Mining Engineering is a bonafide report of the research work carried out by me. The material contained in this Research Thesis has not been submitted to any University or Institution for the award of any degree.

MASOOD

Register No. MN09F01

Department of Mining Engineering,

National Institute of Technology Karnataka, Surathkal.

Place: NITK Surathkal, Srinivasnagar

Date:

CERTIFICATE

This is certify that theResearchThesisentitled"**Design of Portable Rotary Drill Setup and Development of Mathematical Model for Estimating Physico-Mechanical Properties of Igneous Rocks Through Noise Measurement**" submitted by Masood (Registration No. MN09F01) as the record of the research work carriedoutbyhim, is accepted as the Research Thesissubmission in partial fulfillment of the requirementsfortheaward of degree of Doctorof Philosophy.

Research Guides

(Name and Signature with date and seal)

Dr. HarshaVardhan

Dr. M. Aruna

Chairman –DRPC (Signature with date and seal)

ACKNOWLEDGMENT

It's my great pleasure and opportunity to express my sincere thanks to my guide Dr. HarshaVardhan, Associate Professor, Department of Mining Engineering, NITK Surthkal, for his instance encouragement and painstaking efforts all along this research work. It was valuable experience to learn from him both as a teacher and research guide. It was enlightening to work with him, interact and get inspired. I extendmy sincere thanks and I am grateful tohim for everything hehas done during the process of my research work.

I thank Dr. M Aruna, Associate Professor, Department of Mining Engineering, NITK, Surathkal, who also agreed to be my guide and motivated me for successful completion of my research work.

I wish to express my gratitude to Prof. M. Govinda Raj, Head of the Department of Mining Engineering for his continued support throughout my research work.

I wish to express my thanks to National Institute of Technology Karnataka, Surathkal where I have taken up this research work particularly Department of Mining Engineering and for providing me with the laboratory facilities for conducting my experimental work.

I would like to express my sincere thanks to my beloved principal Dr. C. K. Subbaraya, and Professor and Head T N Krishnaiah and Mr H N Suresh from my parent organization for their unconditional support and inspiration during the course of this research investigation.

I would like to express my sincere thanks to the RPAC members Dr. SetharamNayak and Dr. Narendranath, Professors, Department of Civil Engineering and Mechanical Engineering respectively of National Institute of Technology Karnataka, Surathkal.

iii

I also extend my sincere thanks to Sri Kamath (Late) for his invaluable contribution in design and fabrication of the portable hydraulically assisted drill setup which was primarily used in carrying out drilling process. Further, I also extend my sincere thanks to Mr. Chandrahasa Rai, Mr. Srikanth, Mr. Mahabala and Mr. Annie of the Department of Mining Engineering of National Institute of Technology Karnataka, Surathkal, for their valuable support.

I am grateful for the support and inspiration which I got from my parents and other family members without which I certainly would have not been able to complete this research work.

I am thankful to Prof. D. V. Reddy, Department of Civil Engineering of NITK Surathkal for his constant support and encouragement.

Finally I would like to sincerely thank all other individuals who have contributed directly or indirectly in successful completion of this investigation.

MASOOD

Place: N.I.T.K., Surathkal Date:

ABSTRACT

The process of drilling is invariably associated with noise as a by-product. Scientists and engineers worldwide have widely used acoustic signatures from mechanical equipment's to detect fault in a system. However, the application of sound in determining rock properties is limited. In the past few years, research in the area of predicting rock properties using sound level produced during drilling has been reported by various investigators. However, the equipment used by the earlier investigators was a jack hammer drill which itself was a high noise generating unit and it was believed that the noise from the drilling unit itself might be affecting the results of sound level during measurement. Further, the CNC machine used by the earlier investigators was rarely used in mines. Added to that, the cost of the machine was also high.

In view of the above, it is felt necessary to fabricate a low cost and low noise generating drilling unit for the purpose of estimating rock properties. Hence the primary focus of this research work is development and fabrication of a portable, cost effective, rotary drilling set-up for drilling in rocks of varying physico-mechanical properties. Further, development of general prediction mathematical models for different physico-mechanical properties of igneous rock using statistical methods is also one of the component of the research work. The purpose of this research work is to design and fabricate a low cost and low noise producing drilling setup which will be very useful in the estimation of rock properties using sound level produced during drilling. The developed models based on the results of laboratory studies, clearly indicate the prediction of the physico-mechanical properties of igneous rocks with acceptable degree of accuracy using this designed drill setup.

TABLE OF CONTENT

CHAPTER NO	PARTICULARS	PAGE NO
CHAPTER 1	DECLARATION CERTIFICATE ACKNOWLEDGMENT ABSTRACT TABLE OF CONTENTS LIST OF TABLES LIST OF FIGURES NOTATIONS NOMENCLATURE INTRODUCTION	i iii v vi ix xii xiii xiv
		1
CHAPTER 2	LITERATURE REVIEW Determination of Rock Properties through	4
2.1	Experimental Work	4
2.2	Determination of Rock Properties Using Sound Level	6
2.3	Application of Regression Techniques in Geo- mechanics	13
CHAPTER 3 OBJECTIVES AND SCOPE OF THE WORK		
3.1		20
3.2	Scope of the Work	20
CHAPTER 4 4.1 4.1.1 4.1.2 4.2 4.2.1 4.2.1 4.2.2 4.2.3	Equipment's AIM-317E-Mu Compression Testing Machine Brazilian Tensile Testing Machine Instrumentations DIGI – Schmidt 2000 Spark 706 Sound Level Measuring Instrument Porosity	22 22 23 24 24 25 27
4.2.4	Dry Density	27
CHAPTER 5 5.0 5.1 5.1.1 5.1.2 5.1.3 5.2 5.3 5.4 5.5 5.6	DRILL SETUP Introduction Fabrication of Portable Drill Machine The Drilling Unit Water Storage and Supply Unit Hydraulic Sub-Unit Complete Set-up with Different Components Working Principle of Drill Set-Up Precautions to be Observed Cost Considerations and Effectiveness of Drill Set-Up Closure	28 28 29 32 35 35 35 36 39 39

CHAPTER 6	LABORATORY INVESTIGATIONS	41
6.1	Preparation of Test Specimens	41
6.2	Drill Bits Used for Drilling Rock Samples	47
6.3	Determination of Rock Properties	48
6.3.1	Compressive Strength	48
6.3.2	Tensile Strength	48
6.3.3	Schmidt Rebound Number	48
6.3.4	Porosity	49
6.3.5	Dry Density	49
6.3.6	Inference	50
6.4	Drilling: Designed Drilling Set-Up	51
6.4.1	Operating Procedure	52
6.5	Measurement of Sound Levels	53
6.6	Selection of Drill Parameters	53
6.7	Experimental Procedure	54
6.8	Results and Discussion	55
6.8.1	With 16 mm Drill Bit and 15 Kg/cm ² Thrust	55
6.8.2	With 16 mm Drill Bit and 18 Kg/cm ² Thrust	57
6.8.3	With 16 mm Drill Bit and 20 Kg/cm ² Thrust	58
6.8.4	With 16 mm Drill Bit and 25 Kg/cm ² Thrust	60
6.8.5	With 18 mm Drill Bit and 15 Kg/cm ² Thrust	61
6.8.6	With 18 mm Drill Bit and 18 Kg/cm ² Thrust	62
6.8.7	With 18 mm Drill Bit and 20 Kg/cm ² Thrust	64
6.8.8	With 18 mm Drill Bit and 25 Kg/cm ² Thrust	65
6.8.9	With 20 mm Drill Bit and 15 Kg/cm ² Thrust	66
6.8.10	With 20 mm Drill Bit and 18 Kg/cm ² Thrust	68
6.8.11	With 20 mm Drill Bit and 20 Kg/cm ² Thrust	69
6.8.12	With 20 mm Drill Bit and 25 Kg/cm ² Thrust	70
6.9	Inference	
0.5		75
CHAPTER 7	MATHEMATICAL MODELLING	76
7.1	Introduction	77
7.2	Unknown Parameters	77
7.3	Necessary Number of Independent Measurements	79
7.3.1	Statistical Assumptions	79
7.4	Underlying Assumptions	79
	Mathematical Models Using Multiple Regression	
7.5	Analysis	81
7.6	Input to a Regression Problem	83
7.7	Multiple Regression Analysis and ANOVA Technique	85
7.7.1	Selection of Rock Samples for Modelling	87
	Mathematical Regression Model for Different Igneous	
7.8	Rock Properties	88
7.9		
	Performance Prediction of the Derived Models	95
	Performance Prediction of the Derived Models	95
7.10	Performance Prediction of the Derived Models Inference	95 97
7.10 CHAPTER 8		97
7.10 CHAPTER 8 RESEARCH	Inference CONCLUSIONS AND SCOPE FOR FURTHER	97 98
7.10 CHAPTER 8	Inference	97

REFERENCES	100
Appendix – I	111

LIST OF TABLES

TABLE NO	PARTICULARS	PAGE NO
4.1	Specifications of Sound Level Meter	26
6.1	The Average Values of Physico-Mechanical Properties of Igneous Rocks	50
6.2	Equivalent Sound Levels (L_{eq}) with 16 mm Diameter Drill Rod at 15 kg/cm ² Thrust	55
6.3	Equivalent Sound Levels (L_{eq}) with 16 mm Diameter Drill Rod at 18 kg/cm ² Thrust	57
6.4	Equivalent Sound Levels (L_{eq}) with 16 mm Diameter Drill Rod at 20 kg/cm ² Thrust	58
6.5	Equivalent Sound Levels (L_{eq}) with 16 mm Diameter Drill Rod at 25 kg/cm ² Thrust	60
6.6	Equivalent Sound Levels (L_{eq}) with 18 mm Diameter Drill Rod at 15 kg/cm ² Thrust	61
6.7	Equivalent Sound Levels (L_{eq}) with 18 mm Diameter Drill Rod at 18 kg/cm ² Thrust	62
6.8	Equivalent Sound Levels (L_{eq}) with 18 mm Diameter Drill Rod at 20 kg/cm ² Thrust	64
6.9	Equivalent Sound Levels (L_{eq}) with 18 mm Diameter Drill Rod at 25 kg/cm ² Thrust	65
6.10	Equivalent Sound Levels (L_{eq}) with 20 mm Diameter Drill Rod at 15 kg/cm ² Thrust	66
611	Equivalent Sound Levels (L_{eq}) with 20 mm Diameter Drill Rod at 18 kg/cm ² Thrust	68
6.12	Equivalent Sound Levels (L_{eq}) with 20 mm Diameter Drill Rod at 20 kg/cm ² Thrust	69
6.13	Equivalent Sound Levels (L_{eq}) with 20 mm Diameter Drill Rod at 25 kg/cm ² Thrust	70
7.1	Significance of regression coefficients for estimation of uniaxial compressive strength (igneous rocks).	89
7.2	Analysis of variance (ANOVA) for the selected quadratic model for estimation of UCS (igneous rocks).	89

7.3	Model summary for dependent variable (UCS - igneous rocks)	89
7.4	Significance of regression coefficients for estimation of Schmidt rebound number (igneous rocks).	90
7.5	Analysis of variance (ANOVA) for the selected quadratic model for estimation of Schmidt rebound number (igneous rocks).	91
7.6	Model summary for dependent variable (SRN - igneous rocks)	91
7.7	Significance of regression coefficients for estimation of Dry density (igneous rock).	92
7.8	Analysis of variance (ANOVA) for the selected quadratic model for estimation of Dry density (igneous rock).	92
7.9	Model summary for dependent variable (Dry Density – igneous rock)	92
7.10	Significance of regression coefficients for estimation of Tensile strength (igneous rock).	93
7.11	Analysis of variance (ANOVA) for the selected quadratic model for estimation of Tensile strength (igneous rock)	93
7.12	Model summary for dependent variable (Tensile strength - igneous rock).	94
7.13	Significance of regression coefficients for estimation of % Porosity (igneous rock).	94
7.14	Analysis of variance (ANOVA) for the selected quadratic model for estimation of % Porosity (igneous rock).	95
7.15	Model summary for dependent variable (% Porosity - igneous rock).	95
7.16	Performance prediction indices of the regression models (Igneous Rock)	96

LIST OF FIGURES

FIGURE NO	PARTICULARS	PAGE NO
4.1	AIM-317E-Mu Compression Testing Machine	22
4.2	Brazilian Tensile Testing Machine	24
4.3	DIGI-SCHMIDT 2000, SRN Measuring Instrument	25
4.4	Spark 706, Sound Measuring Instrument	26
5.1(a)	Drill Unit with Different Components	30
5.1(b)	Drill Unit with Different Components	31
5.2	Water Storage and Supply Unit for the Experimental Drill Set-Up	33
5.3	Hydraulic Pump of the Experimental Drilling Set-Up	37
5.4	Complete Unit With Different Components	38
6.1	Rock Samples Used for the Investigation	43
6.2	Tungsten Carbide Drill Bits	47
6.3	Sound Level Vs Rock Type for 16 mm Diameter Drill Rod at 15 kg/cm ² Thrust	56
6.4	Sound Level Vs Rock Type for 16 mm Diameter Drill Rod at 18 kg/cm ² Thrust	58
6.5	Sound Level Vs Rock Type for 16 mm Diameter Drill Rod at 20 kg/cm ² Thrust	59
6.6	Sound Level Vs Rock Type for 16 mm Diameter Drill Rod at 25 kg/cm ² Thrust	60
6.7	Sound Level Vs Rock Type for 18 mm Diameter Drill Rod at 15 kg/cm ² Thrust	62
6.8	Sound Level Vs Rock Type for 18 mm Diameter Drill Rod at 18 kg/cm ² Thrust	63
6.9	Sound Level Vs Rock Type for 18 mm Diameter Drill Rod at 20 kg/cm ² Thrust	65

6.10	Sound Level Vs Rock Type for 18 mm Diameter Drill Rod at 25 kg/cm ² Thrust	66
6.11	Sound Level Vs Rock Type for 20 mm Diameter Drill Rod at 15 kg/cm ² Thrust	67
6.12	Sound Level Vs Rock Type for 20 mm Diameter Drill Rod at 18 kg/cm ² Thrust	69
6.13	Sound Level Vs Rock Type for 20 mm Diameter Drill Rod at 20 kg/cm ² Thrust	70
6.14	Sound Level Vs Rock Type for 20 mm Diameter Drill Rod at 25 kg/cm ² Thrust	71
6.15	Sound Level V/s Tensile Strength for 18 mm diameter and 20 kg/cm ² Thrust Pressure	72
6.16	Sound Level V/s UCS for 18 mm diameter and 20 kg/cm ² Thrust Pressure	72
6.17	Sound Level V/s SRN for 18 mm diameter and 20 kg/cm ² Thrust Pressure	73
6.18	Sound Level V/s Density for 18 mm diameter and 20 kg/cm ² Thrust Pressure	73
6.19	Sound Level V/s Porosity for 18 mm diameter and 20 kg/cm ² Thrust Pressure	74
6.20	Sound Level V/s Different Mechanical Properties for 18 mm diameter and 20 kg/cm ² Thrust Pressure	74

NOTATIONS

- L_{eq} Equivalent sound level
- ρ Density
- V_pP-wave velocity
- n Porosity

R²correlation coefficient or regression coefficient

- β Unknown Parameter
- X independent variables
- Y dependent variable
- f Response Variable
- A Drill bit diameter
- B Drill bit speed
- C Thrust
- D A-weighted Equivalent sound level
- k The number of model parameter
- n The Number of scenarios
- x_i Independent process variable
- b_i Linear effect of x_i
- b_{ij} Quadratic Effect of x_i
- ε Fitting error in multiple regression model
- R₂Coefficient of determination

NOMENCLATURE

ISRM	International Society for Rock Mechanics
UCS	Uni-axial Compressive Strength
AE	Acoustic Emission
IJMME	International Journal of Mining & Mineral
Engineering	
RMRE	Rocks Mechanics & Rock Engineering
MLP	Multi Layer Perceptron
RBF	Radial Basis Function
SRN	Schmidt Rebound Number
TS	Tensile Strength
E	Elasticity
ANN	Artificial Neural Network
SV	Sound Velocity
SRN	Schmidt Rebound Number
RPM	Revolution Per Minute
ANOVA	Analysis of Variance
MSE	Mean Square Error
MS	Mean Square
DF	Degrees of Freedom
SS	Sum of Squares
VAF	Values Account For
RMSE	Root Mean Square Error
MAPE	Mean Absolute Percentage Error

CHAPTER 1

1.0 INTRODUCTION

Drilling in rock/strata of any type invariably produces noise as a by-product. Particularly in hard rocks, jackhammer drill is extensively used for the purpose of drilling which has the major noise source as the driving unit emitting high intensity low frequency noise due to compressed air (Powel, 1956). It has been estimated that, of the total noise energy of pneumatic drill, 87.5% is contributed by the exhaust and the next largest component is the impact between the piston and the drill steel (Walker, 1963; Holdo, 1958 & Miller, 1963).

Acoustic signatures from mechanical equipments had been widely used to detect fault in a system. However, the application of sound in determining rock properties was very limited. In the past few years, research in the area of predicting rock properties using sound level produced during drilling had been reported by various investigators.

Earlier, few studies in oil and gas industries seemed to have proposed a technique called "Seismic While Drilling" for estimating rock formations. For instance, few studies had proposed the use of noise produced by the bit during drilling as a seismic source for surveying the area around a well and also for formation characterization while drilling (Onyia, 1988; Martinez, 1991; Rector & Hardage, 1992; Miranda, 1996; Asanuma & Niitsuma, 1996; Hsu, 1997; Aleotti et al, 1999; Tsuru & Kozawa, 1998; Hand et al, 1999; Fernandez & Pixton, 2005). A recent study (Stuart et al, 2007)has also reported a method of estimating formation properties by analyzing acoustic waves that are emitted from and received by a bottom hole assembly. It needs to be emphasized that "Seismic

While Drilling" technique is different from the technique of estimating rock properties using sound levels produced during drilling.

An attempt was made by Vardhan & Murthy (2007) to investigate the influence on sound level due to drilling in rocks of varying physico-mechanical properties using jackhammer drill. The results indicated higher sound level at various measurement locations with increase in compressive strength and with lower values of abrasivity of rocks.Based on the results of Vardhan & Murthy (2007), in the year 2009, Vardhan, Adhikari & Govindaraj indicated that sound level can be a promising tool in estimating rock properties during drilling. A number of investigation were carried out by Vardhan & Yadav (2011); Rajesh Kumar, Vardhan & Govindaraj (2011)^{abc}; Rajesh Kumar, Vardhan, Govindaraj & Vijay (2013); Rajesh Kumar, Vardhan, Govindaraj & Sowmya Saraswathi (2013) to determine the rock properties using sound level produced during drilling with different degrees of success. A number of models both using statistical techniques as well as using Artificial Neural Networks were developed by the above mentioned investigators. However it is important to mention here that the entire investigation by the above mentioned researchers concentrated in using a computer numerical controlled (CNC) vertical milling machine for drilling holes with drill bit diameters ranging from 6 to 20mm with a shank length of 40 mm. This equipment hardly found application in mining industry. Added to that, it was capital intensive.

Kivade et al (2011& 2014^{ab}) studied in detail the effect of applied thrust on the penetration rate and sound level of jackhammer drill. It was said that the compressive strength and abrasivity exhibit strong correlations with the sound level and penetration rate. All the investigation of Kivade et al (2011 & 2014^{ab}) was based on experimental

work in the laboratory using jackhammer drill, which itself was a high noise generating equipment.

Based on the above, it is clear that, there is no doubt regarding estimation of rock properties using sound level produced during drilling. However, the equipment used by the earlier investigators was a jack hammer drill which itself was a high noise generating unit and it was believed that the noise from the drilling unit itself might be affecting the results of sound level during measurement. Further, the CNC machine used by the earlier investigators was rarely used in mines. Added to that, the cost of the machine was also high.

In view of the above, it is necessary to fabricate a low cost and low noise generating drilling unit. It is believed that, such a setup will be more efficient in estimation of rock properties using sound level produced during drilling. Added to that, due to its lower cost, it can be easily used by even small quarry operators who are reluctant in spending more money on purchasing high end equipments.

CHAPTER 2

2.0 LITERATURE REVIEW

2.1DETERMINATION OF ROCK PROPERTIES THROUGH EXPERIMENTAL WORK

Determining and estimating the physico–mechanical properties of rock is one of the important aspects for the mining and civil engineers in the area of rock mechanics. The International Society for Rock Mechanics (ISRM) has suggested a number of methods for measuring rock properties both in the laboratory as well as in the field which were compiled and edited by Ulusay and Hudson (2007).

Schmidt (1951) had designed a portable hammer to conduct non-destructive tests on concrete which was popularly known as Schmidt hammer and was widely used for indirectly estimating the rock strength. The Schmidt hammerbasicallymeasured the surface rebound hardness of the tested material. In the year 2009, Aydin proposed a revised suggested method, which superceded the portion of earlier ISRM document, to determine the rebound hardness of rock surfaces both in laboratory conditions and insitu with an emphasis on the use of this hardness value as an index of the uniaxial compressive strength and Young's modulus of rock materials.

Szlavin (1974) carried out an analysis to find out if statistically significant correlations existbetween the physico-mechanical properties of rock. According to him, this would make it possible for the estimates to be made of one property from any other single property. Various tests such as Compressive Strength, Tensile Strength, Shore Hardness, Indentation, Specific Energy and Abrasivity were conducted on a number of samples and the arithmetic mean value was calculated which was used in the analysis.

A program was devised so that the test results could be fed into a computer. The relationships between the variables were obtained in terms of regression coefficients, standard deviations and correlation coefficients. A comparison of the results showed that the majority of the 'direct' physico-mechanical propertiescould be estimated with reasonable accuracy from each other. However, more errors were involved in the determination of, and calculation from, the energy based units, i.e. specific energy index and abrasivity. Szlavin (1974) also pointed out that, the ratio of uniaxial compressive strength and specific energy was approximately constant.

In the year 1998, Szwedzicki proposed a standard indentation test, as a measure of hardness and its use to predict the uniaxial compressive strength (UCS). According to him, standardized indentation testing allowed for the characterization of physico-mechanical properties of rock. Further, there was a relationship between the value of the indentation hardness index and the UCS.

Gill et al. (2005), by rigorous statistical procedure, proposed an algorithm to determine the minimum number of specimens required in the laboratory testing for the determination of physico-mechanical properties of rocks.

Lama &Vutukuri (1978) and Carmichael (1982) through the results of field investigation, tabulated extensive lists of various mechanical properties of sedimentary rocks from different locations around the world. Kwasniewski (1989) determined the UCS and porosity data of various sandstones. Jizba (1991) presented the physico-mechanical properties of sandstones and shales with a wide range of porosity recovered from various depths in Texas, USA. Wong et al. (1997) also presented a table of strength and other physico-mechanical properties of several representative porous sandstones.

Bradford et al. (1998) and Horsrud (2001) also reported the laboratory test results on the North Sea sandstone and shale, respectively.

2.2 DETERMINATION OF ROCK PROPERTIES USING SOUND LEVEL

Sound has been widely used for fault diagnosis in mechanical industry. It is a very important tool which is even used in predictive maintenance of equipments. However, its application in rock mechanics is very limited. The application of sound in rock mechanics particularly for determination and estimation of rock properties has started very recently in the last few years.

Acoustic emission studies related to rocks were initiated by Obert (1941) and Obert& Duvall (1942) to predict rock bursts in mines. Several investigators (Knill et al. 1968, Hardy 1972, Hardy 1977, Mogi 1973, Shamina 1975, Byerlee 1978) diverted their attention on the change in spectral amplitudes of acoustic waves over a wide band of monitoring frequencies with increasing applied stress.

McNally (1990) proposed an exponential relationship between UCS and the sonic log after study of the UCS test results on thousands of core samples with geophysical logs from Bowen Basin. As reported by Ward (1998), the German Creek Mine derived its own local formula.

Hatherly (2002) proposed an alternative approach to estimate the UCS of elastic rocks from geophysical borehole logs. He first determined the composition of the rocks and then correlated this to the UCS.

Schon (1996) also found that there was either a linear, polynomial or logarithmic relationship between seismic velocity and UCS.

Zborovjan (2001) showed that rock acoustic signature could be found between the frequency ranges 5000 – 8000 Hz. It was also pointed out that the lower half of signal spectrum was mainly due to the noise produced from the drilling engine and cooling water.

In the year 2003, Zborovjan et al. discussed about acoustic identification of rocks during the process of drilling. An attempt was made by him to identify specific acoustic signature of each drilled rock type and the rock class type. A software was developed for acoustic rock identification based on Hidden Markov Models (Rasmussen 2000). The developed software included the real experimental results to identify the specific acoustic signature of each drilled rock type. According to him, the proposed software gave satisfying results in rock class acoustic identification.

Miklusova et al. (2006) developed an experimental setup to simulate rock disintegration by rotary drilling in the laboratory. She discussed about the analysis of noise signals as acoustic behavior of rock drilling process. According to her, acoustic signal had the potential to be used for control of rock disintegration process.

Futo et al. (2003) discussed the possibility of rock drilling optimization through acoustic signals. According to the investigators, the equivalent level for representative frequency depended on the rock types. Hence, equivalent sound levels could be used for identification of the rock type.

According to Gradl et al. (2008), bit characteristics could be determined using acoustical data alone, i.e. noise of a bit. Investigators could detect major differences in the frequency characteristics during the analysis of the roller cone bit and the natural diamond bit.

Williams & Hagan (2006) investigated the changes in rock cutting conditions with the nature of acoustic signals generated in rock. According to them, acoustic emission vary with time during the rock cutting process and there exists a correlation between the rise in cutting force and the levels of acoustic emission.

In 1994, Jung et al. conducted indentation experiments on rock core specimens. They recorded and analyzed Load – displacement, Acoustic Emission (AE), Root Mean Square – time, AE number – time and AE signal – time curves. They concluded that, during indentation, acoustic emission activities increase with increasing stress level. It rises sharply as the critical load was reached and chipping occurred. According to him, the acoustic emission to a large extent is dependent on rock hardness which was closely associated with its mineral composition, structure and texture and physical properties.

An attempt was made by Vardhan& Murthy (2007) to investigate the influence on sound level due to drilling in rocks of varying physical properties i.e. compressive strength and abrasivity using jackhammer drill. For this purpose, a jackhammer drill setup was fabricated wherein the thrust applied can be varied while drilling vertical holes. The results of this study indicated that, increase in thrust increases the sound level at higher midband frequencies in the noise spectrum. Both the thrust and air pressure were found to have a significant effect on the sound level produced by jackhammer drill at all the

measurement locations. It was said that to maintain a constant penetration rate in the rocks, both the thrust and air pressure need to be increased with an increase in compressive strength and decrease in rock abrasivity. Therefore, increased compressive strength and lower abrasivity of rocks will require higher air pressure and thrusts to be applied to achieve an optimum penetration rate and therefore will result in higher sound level at the operator's position and at other measurement locations.

Based on results of Vardhan& Murthy (2007), in the year 2009, Vardhan, Adhikari&Govindaraj indicated that sound level could be a promising tool in estimating rock properties during drilling.

Rajesh Kumar, Vardhan&Govindaraj (2011^a) developed empirical models using multiple regression technique for prediction of physico-mechanical properties of metamorphic rocks. The model considered the parameters as drill bit diameter, bit speed, penetration rate and equivalent sound level produced during drilling. The F-test was used to check the validity of the developed models. The experimentally measured rock property values and the values calculated from the developed regression model were fairly close which indicated that the developed models could be efficiently used in prediction of intact metamorphic rock properties.

Rajesh Kumar, Vardhan&Govindaraj (2011^b) also developed a general prediction model to investigate the relationships between sound level produced during drilling and physico-mechanical properties such as uniaxial compressive strength, tensile strength and percentage porosity of sedimentary rocks. The results were evaluated using the multiple regression analysis taking into account the interaction effects of various predictor variables. Predictor variables selected for the multiple regression model were

the drill bit diameter, drill bit speed, penetration rate and equivalent sound level produced during rotary drilling (L_{eq}). The constructed models were checked using various prediction performance indices. It was said that the constructed models can be used for practical purposes.

Rajesh Kumar, Vardhan&Govindaraj (2011^c) also made an attempt to estimate rock properties such as uniaxial compressive strength, Schmidt rebound number and Young's modulus using sound level produced during rotary drilling. For this purpose, a computer numerical controlled vertical milling centre was used for drilling holes with drill bit diameters ranging from 6 to 20 mm with a shank length of 40 mm. Fourteen different rock types were tested. The study was carried out to develop the empirical relations using multiple regression analysis between sound level produced during drilling and rock properties considering the effects of drill bit diameter, drill bit speed and drill bit penetration rate. The F-test was used to check the validity of the developed models. The measured rock property values and the values calculated from the developed regression model were fairly close, indicating that the developed models could be efficiently used with acceptable accuracy in prediction of rock properties.

Kivade et al (2011) studied in detail the effect of applied thrust on the penetration rate and sound level of jackhammer drill.

Rajesh Kumar, Vardhan, Govindaraj& Vijay (2013) made an attempt to predict rock properties using soft computing techniques such as multiple regression, artificial neural network (MLP and RBF) models, taking drill bit speed, penetration rate, drill bit diameter and equivalent sound level produced during drilling as the input parameters. A database of 448 cases were tested for determination of uniaxial compressive strength (UCS),

Schmidt rebound number (SRN), dry density (r), P-wave velocity (V_p), tensile strength (TS), modulus of elasticity (E) and percentage porosity (n) and the prediction capabilities of the models were then analyzed. Results from the analysis demonstrated that, neural network approach was efficient when compared to statistical analysis in predicting rock properties from the sound level produced during drilling.

Rajesh Kumar, Vardhan, Govindaraj&SowmyaSaraswathi (2013) reported that in many rock engineering applications such as foundations, slopes and tunnels, the intact rock properties are not actually determined by laboratory tests, due to the requirement of high quality core samples and sophisticated test equipments. An artificial neural network models were reported to be developed to predict the rock properties of the intact rock, by using sound level produced during rock drilling. A database of 832 datasets, including drill bit diameter, drill bit speed, penetration rate of the drill bit and equivalent sound level (L_{eq}) produced during for input parameters, and uniaxial compressive strength (UCS), Schmidt rebound number (SRN), dry density (ρ), P-wave velocity (Vp), tensile strength (TS), modulus of elasticity (E) and percentage porosity (n) of intact rock for output, was established. The constructed models were checked using various prediction performance indices. Goodness of the fit measures revealed that recommended ANN model fitted the data as accurately as experimental results, indicating the usefulness of artificial neural networks in predicting rock properties.

Investigation was carried out by Kivade et al (2014) on ten rock samples using pneumatic drill with drill bits of different diameters. The sound level and penetration rates were correlated with the rock properties. It was said that the compressive strength and abrasivity exhibit strong correlations with the sound level and penetration rate. It

was concluded that, among the rock properties included in the study, the compressive strength and abrasivity values were the dominant ones affecting the penetration rate and sound level of percussive drills.

Kivade et al (2014) also reported about a detailed study carried out to determine the influence of rock properties on the sound level produced during pneumatic drilling. Further, investigation was also carried out on the effect of thrust, air pressure and compressive strength on penetration rate and the sound level produced. It was observed that, very low thrust resulted in low penetration rate. Even very high thrust did not produce high penetration rate at higher operating air pressures. With increase in thrust beyond the optimum level, the penetration rate started decreasing and caused the drill bit to 'stall'. Results of the study showed that penetration rate and sound level increased with the increase in the thrust level. After reaching the maximum, they started decreasing despite the increase of thrust. A general prediction model was developed to investigate the relationships between sound level produced during drilling and physical properties such as uniaxial compressive strength and abrasivity of sedimentary rocks. The results were evaluated using the multiple regression analysis taking into account the interaction effects of predictor variables.

SrisharanShreedharan, ChiranthHegde, Sunil Sharma and Vardhan (2014) also reported on method to monitor and evaluate the sounds produced as undesirable byproducts, at the drill-bit and rock interface, to predict the type of rock being drilled. It was said that the method might be extrapolated further in the estimation of rock properties.

2.3 APPLICATION OF REGRESSION TECHNIQUES IN GEO-MECHANICS

There had been a steady increase in the successful and potential application of regression in many areas of geo-mechanics in the recent past.

Sachapazis (1990) carried out tests to determine Schmidt hammer rebound hardness number, Tangent Young's modulus, and uniaxial compressive strength for 29 different types of carbonate rocks. The correlations between these parameters were carried out and regression equations were established among the above said properties with high coefficients of determination (\mathbb{R}^2).

A statistical approach was proposed by Kim &Gao (1995), for the computation of the mechanical properties of rock mass. It was said that the approach accounts for the uncertainty due to the variability of the rock material properties and the pattern of the discontinuities in the rock mass. All parameters describing the rock mass properties were considered random variables instead of a constant. Probability distributions of the deformation modulus and the compressive strength of a rock mass were obtained by applying Monte Carlo simulation. Using extreme value statistics, the input data used for the simulation and the simulation results were analyzed. It was said that the third type asymptotic distribution of the smallest values was general statistical representation of mechanical properties of rocks.

Using Schmidt Hammer rebound readings of seven rock types, Katz et al. (2000) developed new empirical correlations with laboratory measured values of Young's modulus, Uniaxial Compressive Strength and density. According to the investigators,

these developed equations could be used to estimate the relevant mechanical properties in the field as well as laboratory.

An evaluation of the correlations using least squares regression, between the uniaxial compressive strength (UCS) values of 48 different rocks and the corresponding resulted of point load, Schmidt hammer, sound velocity and impact strength tests were carried out by Kahraman in the year 2001. By calculating the coefficient of variation, the variability of test resulted for each test and each rock type was evaluated. According to him, strong linear relations existed between the point load strength index values and UCS values for the coal measure rocks and other rocks. The Schmidt hammer and sound velocity tested exhibit significant nonlinear correlations with the compressive strength of rock. In the sound velocity test, the data points were scattered at higher strength values. There was no clear relation between the impact strength values and compressive strength values for coal measure rocks. A weak nonlinear correlation was found between the impact strength values and the compressive strength values for the other rocks. Based on his investigation, he concluded that, except for the impact strength, all other test methods evaluated in the study provide reliable estimation of compressive strength of rock.

An investigation using regression analysis was carried out by Altindag (2002), to find out the relationship between brittleness and different rock properties i.e., drillability index, Point load and Elastic modulus based on the data of Bilgin et al. (1993) and Kahraman et al. (2000). According to him, there existed a significant correlation between brittleness and the drillability index.

In the year 2002, Kahraman investigated the relationship between the direct and the indirect P-wave velocity values obtained through laboratory experiments.

Significant amount of investigation had been carried out on the usefulness of Schmidt hammer test on different rock types (Deere & Miller, 1966;Aufmuth, 1973; Beverly et al. 1979;Kidybinski, 1980; Singh et al. 1983; Sheorey et al. 1984;Haramy& DeMarco, 1985; Ghose&Chakraborti, 1986; O'Rourke, 1989; Cargill &Shakoor, 1990;Sachpazis, 1990; Kahraman et al. 1996;Tugrul&Zarif, 1999; Katz et al. 2000; Yilmaz &Sendir, 2002). Various empirical equations had been proposed for estimating UCS and Young's modulus of rock from Schmidt hammer rebound number.

An attempt was made by Yasar& Erdogan (2004a) to find out the statistical relationships between Schmidt hammer rebound number and Shore scleroscope hardness values with various physico-mechanical properties of different rocks. According to them, physico-mechanical properties could be estimated using hardness methods. They also pointed out based on their study that porosity increases with decreasing hardness values and strength of the rocks.

Yasar& Erdogan (2004b) also developed the statistical relations using the least square regression method to predict physico-mechanical properties namely UCS, Young's Modulus and Density of Carbonate rocks using Sound Velocity (SV) index value. Based on their study, they concluded that the above said physico-mechanical properties can be estimated from Sound Velocity values using their mathematical equations.

Singh & Rao (2005) pointed out that there was a strong correlation between ultimate strength and the tangent modulus values of the block mass tested in the laboratory for a specific failure mode, when plotted on the Dure-Miller classification chart and it followed an empirical straight line.

Karakus et al. (2005) used the technique of multiple regression modeling to predict elastic properties of intact rocks from index tests.

Chang et al. (2006) reported that, some of the equations developed work reasonably well (for example, strength – porosity relationships for sandstone and shale) to predict rock properties. However, the authors were of the opinion that rock strength variations with individual physical property measurements scatter considerably, indicating that most of the empirical equations were not sufficiently generic to fit all the data published on rock strength and physical properties.

Kahraman&Alber (2006), correlated the electrical resistivity values with experimentally measured physico-mechanical properties such as uniaxial compressive strength, elastic modulus, point load strength, Schmidt hammer value, P-wave velocity, density and porosity values, using least squares regression method. According to the authors:

- UCS and elastic modulus, increases with increasing electrical resistivity.
- There exists a strong logarithmic relationships between UCS and resistivity and between resistivity and elastic modulus.
- Density linearly increases with increasing resistivity.
- Porosity and electrical resistivity are inversely related.

Based on their investigation, the authors concluded that, electrical resistivity could be used as a representative measure of rock properties, particularly for characterizing rocks for which regularly shaped specimen were difficult to obtain.

Karakurt et al. (2013) carried out an experimental and statistical study on noise level generated during rock sawing by circular diamond saw blades. Influence of the operating variables and rock properties on the noise level were investigated and analyzed. Statistical analysis were then employed and models were built for the prediction of noise levels depending on the operating variables and the rock properties. The derived models were validated through statistical tests. It is found that increasing of peripheral speed, traverse speed and cutting depth result in an increase in noise levels. On the other hand, a decreasing trend in noise level was initially observed with increasing the flow rate of cooling fluid. It was said that there is moderate correlations between uniaxial compressive strength, density and noise levels.

Kahraman et al. (2013) made an attempt to determine the Los Angeles abrasion resistance using indirect method for preliminary investigation as the direct method is time consuming, expensive, and requires a large amount of samples. In this investigation, Los Angeles abrasion, noise level measurement, density and porosity tests were carried out on 27 different rock types including igneous, metamorphic and sedimentary rocks. The test results were evaluated using the simple and multiple regression analysis. A good corelation was found by the investigators between the Los Angeles abrasion loss and the noise level. In order to check the possibility of obtaining more significant relations, multiple regression analysis was performed by including density and porosity values. However, the regression analysis showed that the

correlation coefficients of multiple regression equations were slightly higher than that of the simple regression equation. Since the simple regression equation is practical and statistically significant, it was suggested for estimation purpose. The investigators concluded that Los Angeles abrasion loss of aggregates can be reliably estimated from noise level measurement.

Delibalta et al. (2015) carried out an investigation on the usability of noise level from rock cutting for the prediction of physico-mechanical properties of rocks. According to the investigators, as the indirect tests are easier and cheaper than the direct tests, the prediction of rock properties from the indirect testing methods is important especially for the preliminary investigations. In this study, the predictability of the physico-mechanical rock properties from the noise level measured during cutting rock with diamond saw was investigated. Noise measurement test, uniaxial compressive strength (UCS) test, Brazilian tensile strength (BTS) test, point load strength (Is) test, density test, and porosity test were carried out on 54 different rock types in the laboratory. The results were statistically analyzed to derive estimation equations. Strong correlations between the noise level and the mechanical rock properties were found. The relations follow power functions. Increasing rock strength increases the noise level. Density and porosity also correlated strongly with the noise level. The relations follow linear functions. Increasing density increases the noise level while increasing porosity decreases the noise level. It was said that, the developed equations are valid for the rocks with a compressive strength below 150 MPa. Finally, it was concluded that the physico-mechanical rock properties can reliably be estimated from the noise level measured during cutting the rock with diamond saw.

Faisal et al. (2007), carried out linear statistical analysis to establish the relationships between engineering properties of the intact rock namely dolomite, dolomitic limestone and shale rocks with different types of hardness i.e., Schmidt, shore scleroscope, abrasion, and total hardness. According to the authors, there existed a good relationship between the engineering properties of the intact rock and its hardness.

Yavuz et al. (2008), investigated the dependence of abrasion rate on physicomechanical properties of rocks namely bulk density, effective porosity, P-Wave velocity, Schmidt rebound hardness, Compressive strength, Tensile strength through least square regression analysis. Based on their investigation, the authors concluded that:

- Hardness, tensile strength, compressive strength and density of rock could adequately estimate the abrasion rate of rocks.
- There is a possibility of rough estimateofabrasion rate of rocks using porosity and P-wave velocity of rocks.
- More abrasion resistant rocks are likely to have high bulk density, compressive strength, tensile strength, hardness and lower value of porosity.

Kilic&Teymen (2008) showed satisfactory correlationsforestimating the physicomechanical properties of rocks using non-destructive and indirect test methods.

CHAPTER 3

3.0 OBJECTIVES & SCOPE OF THE WORK

3.1 OBJECTIVES

Based on the literature review, the OBJECTIVES of the present investigation are as follows:

1. Development and fabrication of a portable, cost effective, rotary drilling set-up for drilling in rocks of varying physico-mechanical properties.

2. Development of general prediction mathematical models for different physicomechanical properties of igneous rock using statistical methods.

3.2 SCOPE OF THE WORK

Based on the suggestions received from the members of the Research Progress Assessment Committee, the SCOPE of the WORK is limited to:

a. Design and fabrication of the drilling setup.

b. Development of general prediction mathematical models for different physicomechanical properties of igneous rock through multiple regression analysis in order to establish the relationship between:

- i. Sound levels produced during drilling and uni-axial compressive strength of igneous rocks.
- ii. Sound levels produced during drilling and the tensile strength of igneous rocks.

- iii. Sound levels produced during drilling and the Schmidt Rebound Number.
- iv. Sound levels produced during drilling and the rock density.
- v. Sound level produced during drilling and the porosity of igneous rocks.

CHAPTER 4

4.0 EQUIPMENTS AND INSTRUMENTATIONS

The equipments and instruments described below in this chapter were used to measure various physico-mechanical properties in the laboratory.

4.1 EQUIPMENTS

4.1.1 AIM-317E-Mu Compression Testing Machine

Compressive strength was one of the most important physico-mechanical properties of rock material, used in excavation projects. AIM–317E–Mu micro controller compression testing machine was used for measurement of Uniaxial Compressive Strength (UCS) of rock materials (Fig.4.1).



Fig.4.1 AIM-317E-Mu, Compression Testing Machine

It had an intelligent pace rate controller, motorized pumping unit and loading unit with maximum loading capacity of 2000 kN.

4.1.2 Brazilian Tensile Testing Machine

Rock material generally has a low tensile strength. The low tensile strength is due to the existence of micro cracks in the rock. The existence of micro cracks may also be the cause of rock failing suddenly in tension with a small strain. Tensile strength of rock was measured from Brazilian test loading frame with 100 kN capacity, having a base and a cross head joined together with two solid pillars with nuts (Fig. 4.2).

At the top, the pillars have long threads for height adjustment and on the base, a 100 kN hydraulic jack is centrally fixed between the pillars. This jack has an integral pumping unit and an oil reservoir. A 100 kN capacity pressure gauge is fixed to the jack for indicating the load on the specimen and also an operating handle is provided with the jack.



Fig 4.2 Brazilian Tensile Testing Machine

4.2 INSTRUMENTATIONS

4.2.1 DIGI – Schmidt 2000

Schmidt hammer rebound hardness is often measured during the early part of field investigation. It is a measure of the hardness of the rock material by counting the rebound degree. Tests were performed with DIGI – SCHMIDT 2000 (Fig.4.3). The graphic LCD 128 x 128 pixel display unit of the instrument immediately displays the rebound value.



Fig.4.3 DIGI-SCHMIDT 2000, SRN measuring instrument

4.2.2 SPARK 706 Sound Level Measuring Instrument

The instrument used for sound measurement was a Spark 706 from Larson Davis, Inc., USA (Fig.4.4). The instrument was equipped with a detachable 10.6 mm microphone and 7.6 cm cylindrical mast type preamplifier. The microphone and preamplifier assembly were connected by an integrated 1.0 m cable. A Larson Davis CAL 200 Precision Acoustic Calibrator was used for calibrating the sound level meter. Before taking any measurement, the acoustical sensitivity of the sound level meter was checked using the calibrator. Table 4.1 gives the technical specifications of the sound level meter used in the investigation.



Fig. 4.4 Spark 706, Sound Measuring Instrument

Specification	705 / 705+	All Other			
Measurements Range	40 to 143 dB(RMS) : 801 to 46dB (Pea) Typical				
Max. Pack Level	146 dB SPL				
Detectors	Slow Fa	ist Peak			
Frequency Weighting	A/C un wei	ghted Peak			
Microphone	3/8" El	ectrets			
Memory	4MB Non volatile				
Clock / Calendar	Month, Day, Year , Hr: Min : Sec				
Power Supply	Internal One AA battery	Internal Two AA battery			
	35 Hour Operation	100 Hour Operation			
Dimension	3.9 L x 2.9 W x 0.78 D	5.5 L x 2.5W x 1.25 D			
Weight	7.0 oz (198g)	8.4 oz (238g)			
Standard Met	ANSI S1.4 -1983, ANSI S1.25 -1991, IEC 60651				
	1993, IEC 60804 1993, IEC 61252 1993,				
	Compliant				
Intrinsic Safety	ANSI/UL 913, ANSI/UL 2279, part 11				
Standard Met	CSA 157 – CSA E70 11 (Canada)				
	IEC 60079-11 EN50020 (Europe)				

4.2.3 Porosity

Porosity describes how densely the material is packed. To determine the porosity of the rock samples, the specimens were prepared and tested in accordance with ISRM suggestions (Ulusay and Hudson 2007). Porosity was measured by crushing the rock to fine powder and measuring the volume of powder by displacement of the fluid in a pycnometer. The total volume of pores is calculated as the difference between the volume of the specimen and that of the crushed particle. At least five samples of each rock type were used for measuring porosity.

4.2.4 Dry Density

Density is a measure of mass per unit volume. Density of rock material varies, and is often related to the porosity of the rock. It is sometimes defined by unit weight and specific gravity. The density of each core sample was measured after the removal of moisture from it.

CHAPTER 5

DESIGN AND FABRICATION OF THE EXPERIMENTAL SET-UP

5.0 INTRODUCTION

A number of studies have been reported recently on the application of sound level, which has been discussed elaborately in the literature. All these studies have concentrated on using either CNC or jack hammer machine for the purpose of drilling. CNC machine is widely used in the mechanical industry whereas jack hammer drill is widely used in hard rock mining. Though CNC machine produces less noise, it is a costly equipment and not every small mine/quarry operator will like to procure it. Jack hammer drill is a highly noise producing machine and it is felt that the noise of the jack hammer drill will interfere with noise produced during the process of rock drilling.

In view of the above, it is felt to fabricate a low cost and silent drill machine for drilling in rock blocks to indirectly estimate the physico-mechanical properties of the rocks.

5.1 FABRICATION OF PORTABLE DRILL MACHINE

The entire set-up is fabricated for the purpose of experimental investigation. Basically, the set-up which is portable and noiseless unit in itself consists of three important parts as mentioned below.

- i. Drilling unit
- ii. Water supply and storage unit
- iii. Hydraulic pump unit

5.1.1 The Drilling Unit

The drilling unit with different components of the drill set-up is shown in {Fig.5.1(a) and Fig. 5.1(b)} which consists of the following parts:

1. Metal base	11. Balancing rope
2. Metal structure support	12. Wooden base
3. Column with movable piston	13. Sample holder
4. Box girder	14. Motor regulator
5. 1 HP motor	15. RPM display
6. Belt	16. Two pipes
7. Drive pulleys	17. Cylinder
8. Drill chuck	18. Power cord and
9. Drill rod	19. Two pulley wheels
10. Dead weight	

The drill unit is supported on a strong and rigid metal base, which is most commonly used in drilling machines. The loading structure is designed and fabricated such that not only it withstand the weight of the machine, but also strong enough for cyclic loading during drilling. Further, the metal base is connected to a solid rigid structure for accurate and fast drilling of the collected rock samples. The drill machine is equipped with 1 HP noiseless motor which transmits the power through a belt pulley arrangement; the arrangement is such that the transmission loss is negligible. The speed of the motor can be easily monitored using a motor regulator knob provided just beside the motor assembly. The speed of the drill machine in RPM is displayed by a digital tachometer provided near the speed regulator knob.

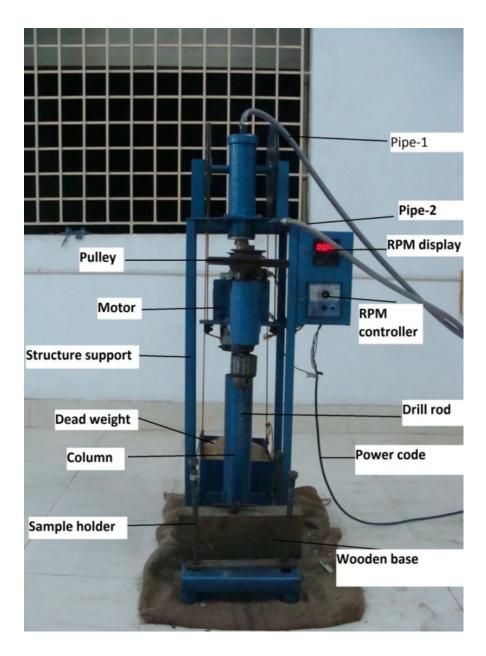


Fig.5.1(a) Drill unit with different components

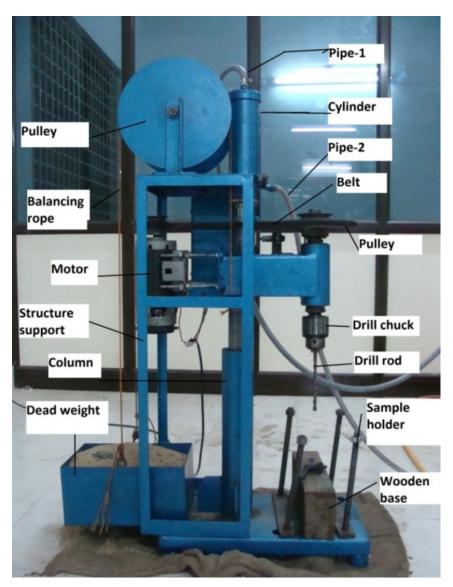


Fig.5.1(b) Drill unit with different components

To hold the drill bit used in the present investigation a chuck with a specialised clamp is used to hold the object firmly during the process of drilling. A two pulley wheel has been equipped to transmit the power. The drive element of a pulley system is belt that runs over the pulley inside the groove. For accurate holding of the work piece, a sample holder with a bolt nut arrangement is provided, such that the rock samples with different sizes can be placed and changed depending upon the length of the bolt to ensure that drilling takes place within few rotations of the drill bit as soon as it comes in contact with the surface of the rock sample. To facilitate the upward and downward moment of the drill bit, a reciprocating piston is provided which reciprocates inside the cylinder which is the central working part of the drill unit. It ascends and descends accordingly with respect to the applied thrust.

5.1.2 Water Storage and Supply Unit

Fig.5.2 shows the different parts of the water storage and supply unit. It consists of Air Vessel, Pressure Vessel, Water Tub, Water Pressure Gauge, 3 Release Pipes, Main Valve, 2 Supply Valves, 2 Release Valves, Feed Pipe, Release Pipe (From Pump)andsuction pipe.

To reduce acceleration heads, air vessels are used on both suction and delivery pipes as for the satisfactory working of a reciprocating pump. The pressure inside the cylinder at any instant must not be less than the vapour pressure of the liquid. In this unit, a pressure vessel holds the liquid at a pressure substantially different from the ambient pressure. If the pressure inside the cylinder is less than or equal to vapour pressure of the fluid, then separation will occur. There are two situations of the piston where this (separation) can happen. One is at the beginning of the suction stroke and the other is at the end of the delivery stroke. Maximum speed in the case of reciprocating pump is determined based on above mentioned condition, i.e., pressure inside the cylinder during suction and delivery stroke should not fall below vapour pressure of the flowing fluid in the suction and delivery pipe.

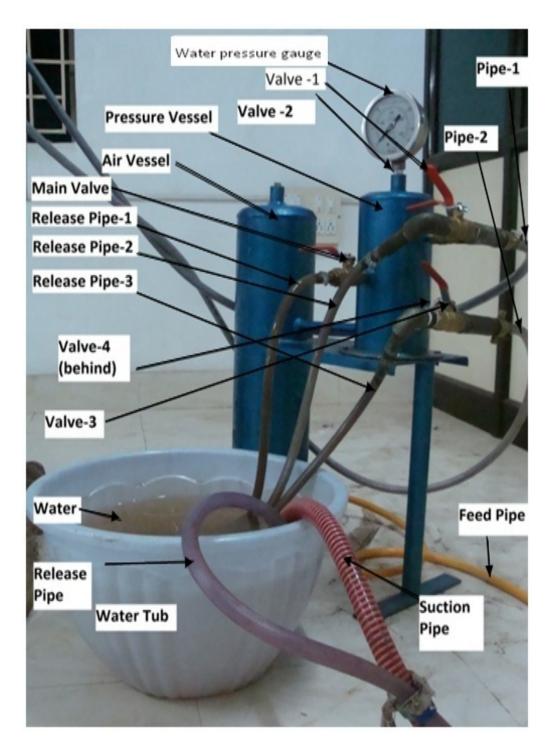


Fig.5.2 Water storage and supply unit for the experimental drill set-up

The pressure at which separation takes place is known as separation pressure and the head corresponding to separation pressure is called separation pressure head.Since an air vessel is a closed chamber (cast iron closed chamber) having an opening at its base, through which water flows into the vessel, or from the vessel, and fitted on the suction as well as on the delivery side, near the pump cylinder to reduce the accelerating head.

Development of acceleration head in the reciprocating pump is undesirable, since it becomes an extra head against which the pump has to work. It is also known that higher the speed and longer the pipe, higher is the acceleration head. However, there is a limit to the speed with which the pump may work from the cavitation close to the cylinder as possible. The vessel is fitted with compressed air, which can contract or expand to absorb most of the pressure fluctuations. An air vessel in a reciprocating pump acts like a flywheel of an engine. Whenever, the pressure rises, water in excess of the average discharge is forced into the air vessel. As the level of the liquid in the air vessel rises, the air held in air chamber gets compressed. When the water pressure in the pipe falls, the compressed air ejects the excess water out. These vessels are capable of absorbing fluctuations in pressure or velocity. It is assumed that the velocity in suction and delivery pipes between air vessels and the cylinder isfluctuating and there is a uniform velocity in pipes beyond the air vessels. When the mean velocity of water in the suction pipes is less than the instantaneous velocity of water in the suction pipe between the air vessel and the cylinder, the required excess water goes out of the air vessel to the cylinder, and when the mean velocity is more than the instantaneous velocity, the excess water goes into the air

vessel. Similarly, for the delivery side, when the mean velocity of water is less than the instantaneous velocity the excess water goes into the air vessel and vice versa.

5.1.3 Hydraulic Sub Unit

Fig.5.3 shows the different parts of the hydraulic pump assembly, which pumps water from the water storage and feeds to the supply unit, which is used by the drilling unit for applying thrust which can be controlled manually. Further, this sub unit is one of the most important part of the drill set-up which consists of different components which are given below:

- i. Motor
- ii. Belt
- iii. Drive Wheel
- iv. Single Piston Pump
- v. Suction Pipe
- vi. Pump Head
- vii. Feed Pipe
- viii. Regulator Knob
- ix. Power cord
- x. Release pipe

5.2 COMPLETE SET-UP WITH DIFFERENT COMPONENTS

The whole set-up as a single assembled one with different sub components is as shown in Fig. 5.4.

5.3 WORKING PRINCIPLE OF DRILL SET-UP

The pump and the motor operations are parallel and not dependent on each other.

The working procedure for the experimental drilling set-up is as follows:

- i. Open all the valves of the pressure vessel once and close the supply and release valves.
- ii. Fit the drill rod to the drill chuck.
- iii. Place the sample on the wooden base and clamp the sample using the sample holder.
- iv. Switch on the motor and pump.
- v. Set the pressure using the main valve and the RPM using the regulator provided.
- vi. Now open the Valve-3 (Release valve) and then Valve -1 (Supply valve).
- vii. Now the piston in the cylinder moves down thus moving the Girder down.
- viii. Thus the drill rod comes in contact with the sample and drills the rock block.

5.4 PRECAUTIONS TO BE OBSERVED

- i. The drill bit should not get red hot during drilling.
- ii. Keep the hands away from the drill bit and belt during operation.
- iii. Use ear protective equipment during drilling the rock sample.
- iv. The pressure should not exceed the maximum value of the pressure gauge.
- v. There should not be any leakage of water through pipes and inter connecting systems.
- vi. Oil level of the water pump should be maintained to its level.
- vii. There should not be any leakage of oil in the water pump.
- viii. The water used should be clean and without solid dirt.
- ix. Sample should be fixed tightly on the wooden base by sample holder.

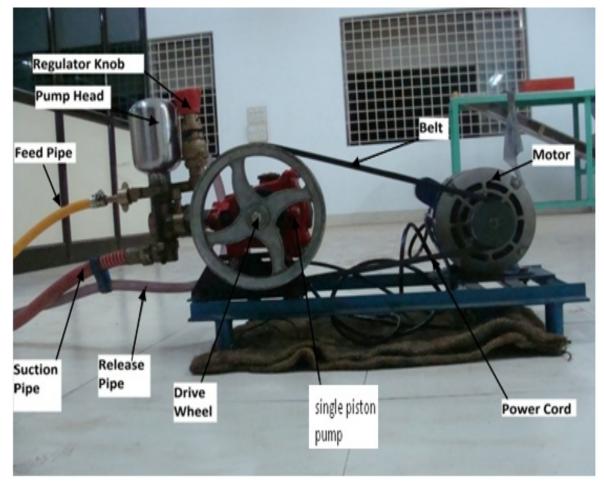


Fig.5.3 Hydraulic pump of the experimental drilling set-up



Fig.5.4 Complete unit with different components

5.5 COST CONSIDERATIONS AND EFFECTIVENESS OF DRILL SETUP

The drilling setup which has been fabricated is an inexpensive and portable device. The cost of this drilling set up is not at all significant compared to other equipment like CNC or Jack hammer drill which has been used by other investigators in the recent past. The overall cost of the complete set-up is only Rs. 55,000=00 which is comparatively less than both CNC machine and that of a jack hammer drill set up. Further, the noise emission from this drilling setup is very low (only of the order of 76dB (A)), thereby making it more suitable for this research work. Both the thrust and RPM on this drill set up can be easily controlled making it very suitable for field applications. Hence it can be anticipated that this set up will be a possible alternative for estimation of physico-mechanical properties of igneous rock samples using sound level produced during drilling in the field.

5.6 CLOSURE

The fabricated drill set-up which produces significantly lesser noise and which is also portable, has been discussed elaborately in this chapter with adequate emphasis made on each and every component as far as the description and the working procedure is concerned. Hence the set-up will be useful in the determination of sound level produced during drilling considering the effects of drill bit diameter, and the amount of thrust applied. The results of laboratory study could be used to predict the physico-mechanical properties of rocks. Since the proposed method is considerably simple and requires lesser time, it may be an economically feasible and a better alternative method in saving the duration of time to conduct experimental investigation and could find application in various engineering applications particularly in geo-technical field.

CHAPTER 6

LABORATORY INVESTIGATION

As discussed earlier, the aim of this work is to develop prediction models for the determination of various properties of igneous rocks from sound levels produced during rotary drilling. Hence, for the purpose of this investigation, twelve different types of igneous rocks were collected from different parts of South India. During sample collection, each block was inspected for macroscopic defects so that it would provide test specimens which are free from any irregularities and defects.

6.1PREPARATION OF TEST SPECIMENS

Twelve different types of igneous rock samples were collected from different locations spread across south India. The collected rock samples were prepared to a volumetric size of 15 cm3. Further, these samples were macro examined to ensure that they are free from any macroscopic defects. Figure 6.1 shows typical sized specimens of igneous rocks used in the present study.





6.1(b)













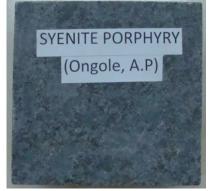




























Fig. 6.1 (a) – 6.1 (I) Rock samples used for the investigation

Below is the brief explanation for twelve types of rocks used in the study:

i. Granite grey is grayish granite formed as intrusive igneous rock with medium to coarse grained texture. The word is derived from Granum (Latin) means grains. The composition consists of quartz and feldspars. The other accessory minerals are mica, apatite and magnetite.

The texture will be medium grained, polycrystalline and equi-granular and the origin is plutonic.

ii. Aptite (Anantapur) is the name given to intrusive rock in which quartz and feldspar are the dominant minerals. Aptites are usually very fine-grained, white, grey or pinkish in colour and their constituents are visible only with the help of a magnifying lens. This type of rock blocks are found at Anantapur in Andhra Pradesh state.

These are igneous rocks with plutonic origin. Fine grained texture is only the distinguishing character because the mineral composition is as same as granite. These rocks are basically formed from magma composition.

iii. Felsite Mysore (also called Felstone) is a very fine grained volcanic rock that may or may not contain larger crystals. Felsite is a field term for a light coloured rock that typically requires petrographic examination or chemical analysis for more precise definition. This rock is typically of volcanic origin, and may be found in association with obsidian and rhyolite (Hosrud 2001). In some cases, it is sufficiently fine-grained for use in making stone tools.

These types of rocks are intermediate ones, where porphyritic texture can be observed clearly.

iv. Gabbro Greenish refers to a large group of dark, coarse-grained, intrusive mafic igneous rocks chemically equivalent to basalt. The rocks are plutonic, formed when molten magma is trapped beneath the earth's surface and cools into a crystalline mass. The vast majority of the earth's surface is underlain by gabbro within the oceanic crust, produced by basalt magmatism at mid-ocean ridges.

It is a plutonic in nature, where the essential mineral is plagioclase and accessory minerals are mica and hypersthenes. Texture is coarse grained, where some varieties shows porphyrytic texture. The compositions of these types of rocks are hornblende, diorite olivine and pyroxenes.

v. Granite Pink (Ilakal) is a common, coarse-grained, light-colored, hard igneous rock consisting chiefly of quartz, orthoclase or microcline and mica.

According to color index pink granite may be leucocratic (light colored) minerals.

vi. Syenite is a coarse-grained intrusive igneous rock of the same general composition as granite but with the quartz either absent or present in relatively small amounts (<5

%). The feldspar component of syenite is predominantly alkaline in character (usually orthoclase). Plagioclase feldspars may be present in small quantities, less than 10%. This type of rock exclusively available in Idapadi of Tamilnadu State.

Further compositions such as alkali feldspars, orthoclase albrite is found in dark coloured minerals such as brotite, hornblende and augite. The accessory minerals are ziricon, sphene.

vii. Granite Porphyry is a common andwidely occurring type of intrusive, felsic, igneous rock which is granular and crystalline in texture. This rock consists mainly of quartz, mica and feldspar. A granitic rock with a porphyritic texture is sometimes known as porphyry.

These types of rocks are plutonic in origin and show porphyrytic texture where it is characterized by the presence of conspicuously large sized crystals known as phenocrysts which are embedded in a fine grained ground mass. These phenocrysts occur at moderate temperature level or intermediate level with in the earth's crusts. It is neither because of rapid cooling nor slow cooling of magma.

viii. Basalt (Nagpur) is a common extrusive volcanic rock. It is usually grey to black in colour and fine-grained due to the fast cooling of lava at the surface of a moon or planet. It may be porphyritic containing larger crystals in a fine matrix, or vesicular, or frothy scoria. Unweathered basalt is black or grey in colour. By definition, basalt is defined as an aphanitic igneous rock that contains, by volume, less than 20 % quartz and less than 10 % feldspathoid and where at least 65 % of the feldspar is in the form of plagioclase (Kahramann, 2001). This type of rocks is found in Nagpur situated in

Maharashtra State. Also, in some of the Deccan traps basaltic rocks are extensively found in states such as Gujrat and Madhya Pradesh.

In these types of rocks we cannot observe the clear crystallization of minerals.

ix. Syenite Porphyry is a very fine grained volcanic rock containing large crystals of phenocrysts which makes it porphyry.

These types of rocks are plutonic in origin and show porphyrytic texture where it is characterized by the presence of conspicuously large sized crystals known as phenocrysts which are embedded in a fine grained ground mass. These phenocrysts occur at moderate temperature level or intermediate level with in the earth's crusts. It is neither because of rapid cooling nor slow cooling of magma.

x. Diorite Porphyry is a grey to dark grey intermediate intrusive igneous rock composed principally of plagioclase feldspar (typically andesine), biotite, hornblende, and/or pyroxene. It may contain small amounts of quartz, microcline and olivine. Zircon, apatite, sphene, magnetite, ilmenite and sulfides occur as accessory minerals (Katz et al., 2001). It can also be black or bluish-grey and frequently has a greenish colour.

xi. Granite (Karnataka) is intrusive, felsic, igneous rock which is granular and crystalline in texture, found in the state of Karnataka. This rock consists mainly of quartz, mica and feldspar. This type of rock blocks are found at Maddur in Karnataka State.

xii. Gabbro (Maddur) is a dark, coarse grained, intrusive igneous rock type. The rocks are plutonic, formed when molten magma is trapped beneath the earth's surface and cools into a crystalline mass.

6.2 DRILL BITS USED FOR DRILLING ROCK SAMPLES



Fig. 6.2 Tungsten carbide drill bits

The term drill may refer to either a drilling machine or a drill bit for use in a drilling machine. In this thesis, for clarity, the term drill bit or bit is used to refer to a bit which is used in a drilling machine and drill refers always to a drilling machine. The drill bit used is made up of tungsten carbide. In the present study drill bits of three different diameters were used i.e. 16 mm, 18 mm and 20 mm. One drill bit was used for drilling six holes and each block facilitates six holes. Drill bits are cutting tools used to create cylindrical holes, almost always of circular cross-section. Drill bits are available in many standard sizes and have many uses (Miklusova et al, 2006). Bits are held in a tool called a drill, which rotates them and provides torgue and axial force to create the hole. Specialized bits are also available for non-cylindrical-shaped holes. The shank is the part of the drill bit grasped by the chuck of a drill. The cutting edges of the drill bit are at one end, and the shank is at the other. In the present investigation drill bits made up of tungsten carbide were used. Tungsten carbide bits posses' typically high resistance to wear and more importantly can be used at high temperatures, thereby preventing the tip of the drill bit getting extremely hot during the course of drilling.

6.3 DETERMINATION OF ROCK PROPERTIES

6.3.1 Compressive strength

For the measurement of uni-axial compressive strength micro controller compression testing machine was used, the details of which was explained in the Chapter V. The igneous rock blocks were prepared with NX-size core specifications, having a length-to-diameter ratio of 2.5:1 as per ISRM standards (Ulusay and Hudson, 2007). Each type of rock was represented by five core specimens. The oven-dried and NX-size core specimens were tested for its compression strength and the average compressive strength for different igneous rock blocks was determined, measured, which are shown in the Table 6.1.

6.3.2 Tensile strength

The rock samples of 54 mm diameter NX-size core specimens, having a length less than 27 mm were prepared as per ISRM standards (Ulusay and Hudson, 2007). The cylindrical surfaces were made free from any irregularities across the thickness using polishing machine. End faces were made flat to within 0.25 mm. The specimen was wrapped around its periphery with one layer of the masking tape and loaded into the Brazilian Tensile Test apparatus across its diameter. Load is applied continuously at a constant rate such that failure occurs within 15-30 seconds. Five sample specimens were tested and the average value was determined and tabulated as shown in the Table 6.1.

6.3.3 Schmidt rebound number

The SRN (Schmidt Rebound Number) was determined using Schmidt Hammer. All tests were performed with the hammer held vertically downwards and at right angles to the

horizontal rock face (Aydin, 2009). To get schmidt hammer rebound number, ten readings were taken on a particular rock sample and then the mean of nearer values were used for the analysis. The basic statistical evaluations of SRN of different igneous rocks are given in the respective Table 6.1.

6.3.4 Porosity

To determine the porosity of the rock samples, the specimens were prepared and tested in accordance with ISRM guidelines (Ulusay and Hudson, 2007). Porosity of rocks under study is given in Table 6.1.

6.3.5 Dry density

Density is a measure of mass per unit of volume. Density of rock material varies and is often related to the porosity of the rock. It is sometimes defined by unit weight and specific gravity. The density of each core sample was measured after the removal of moisture from it. The moisture was removed by placing the samples in an electric oven at approximately 80° C for one hour and then drying at room temperature(Ulusay and Hudson, 2007). The density of dry sample was calculated using equation (6.1).

Dry density, $\rho = \frac{\text{Mass of the sample}}{\text{Volume}} = g/\text{CC}$ (6.1)

Each test was repeated five times and the average dry density of each rock samples were calculated, which is given in Table 6.1.

SI. No.	lgneous rock sample	Sound Level (dB)	Tensile UCS strength(Mpa) (MPa)		SRN	Density (gm/cc)	Porosity (%)
1	Granite Grey	97.55	5.23	46.23	39	2.39	1.73
2	AptiteAnathpur	98.71	5.32	46.50	42	2.40	1.62
3	Felsite Mysore	101.45	5.52	47.60	43	2.41	1.56
4	Gabbro Greenish	102.83	5.70	47.80	47	2.43	1.37
5	Granite Pink Mysore	102.71	5.93	48.0	48	2.50	1.33
6	Syenite	102.75	5.95	48.1	51	2.51	1.33
7	Granite Porphyry	106.55	6.34	51.7	57	2.53	1.20
8	Basalt Nagpur	109.83	6.73	53.2	60	2.56	1.15
9	Syenite Porphyry	111.7	6.81	53.9	62	2.57	0.92
10	Diorite Porphyry	113.88	6.95	57.9	65	2.61	0.83
11	Granite Karnataka	117.31	9.30	77.9	72	2.91	0.56
12	Gabbro Madduru	119.63	12.3	102.6	78	3.30	0.25

Table-6.1: The average values of physico-mechanical properties of igneous rocks

6.3.6 Inference

From Table 6.1 it can be observed that the physico- mechanical properties of igneous rock blocks, increases non linearly with Granite Grey measuring the lowest value of Uni – axial compressive strength (UCS) i.e., 46.23 MPa, whereas, it was found that Gabbro Madduru having the maximum value of 102.6 MPa. Also, the Tensile Strength of different igneous rock blocks considered in the present investigation measures the lowest value of 5.23 MPa for Granite Grey and 12.3 MPa for GrabbroMadduru. Further, the values of SRN and Density measurements are 39 and 2.39 gm/cc for Granite Grey, where as 78 and 3.30 gm/cc for Gabbro Madduru respectively. The values of Porosity decreases from 1.73% to 0.25% for different igneous rock blocks.

It is also significant to mention that the sound level of 97.55 dB(A) was measured with respect to Granite Grey, Where as for Gabbro Madduru was measured with the value of 119.63 dB(A). Hence, it is significant to mention that there is a total difference of 22.08 dB (A) between the rock blocks with lowest UCS of 46.23 MPa and 102.6 MPa of highest UCS. Hence, it can be concluded that rock blocks having highest Tensile Strength, Uni-axial Compressive Strength and Schimdt Rebound Number values generate more degree of sound compared with rock blocks having low physicmechanical property, irrespective of the drill bit parameters used such as dill bit diameter, thrust etc., in the present investigation. Further, it can also be noticed there is very slight variation of sound level of the order of 0.05 dB (A) between Granite Pink Mysore and Syenite rock blocks. Whereas for Granite Karnataka and Diorite Porphyry there is the maximum difference of sound level measurement of the order of 3.43 dB(A). Also, Gabbro Madduru with highest UCS of 102.6 MPa is having sound level of 2.32 dB (A) greater than Granite Karnataka, where both these type of rock blocks are available in Karnataka state. Hence it can be suggested that these two rock blocks may be useful in different engineering applications, where high amount of hardness and strength is necessary.

6.4 OPERATING PROCEDURE FOR THE DRILLING SET-UP

The construction and working principle of drill set has been explained in Chapter IV. The drill consists of one hydraulic pump and a motor which operates in parallel and not dependent on each other. The drill machine was operated in the following sequence:

 Initially all the valves of the pressure vessel were opened, and supply and release valves were closed.

- The drill rod was attached to the drill chuck.
- The rock sample was placed on the wooden base and it is clamped using the sample holder.
- The motor and the pump were switched on.
- The thrust was controlled using the main valve and the RPM was regulated using a regulator.
- Once the required thrust and RPM have been attained, release valve (i.e. valve-3) and supply valve (i.e. valve-1) are opened.
- By doing this piston in the cylinder descends down thus moving the Girder down.
- Now drill rod will come in contact with the surface of the rock sample and drills the rock.

6.4.1 Precautions adopted while drilling

- Turning of the drill bit getting red hot was avoided.
- Ear plugs were used while drilling.
- It was ensured that pressure was not exceeded beyond the maximum value of the pressure gauge.
- Leakage of water through pipes and inter connecting systems was avoided.
- Oil level of the water pump was maintained to appropriate level.
- Leakage of oil in the water pump was avoided.
- It was ensured the water used is clean and free from solid dirt.
- The rock block was fixed firmly on the wooden base using sample holder to avoid slipping of rock block during drilling operation.

6.5 MEASUREMENT OF SOUND LEVELS

For measurement of sound levels 'Spark 706' (from Larson Davis, Inc., USA) dosimeter was used. The microphone of the dosimeter was held at a distance of 1.5 cm from the drill rod, near the rock surface. Cushioning was provided for the drill machine to prevent any noise generation from the floor due to the vibration of drill machine.

Before starting the experiment, sound level was measured six times to ensure whether there is any significant variation in the noise generated from the drill machine. The noise level measurements thus taken indicated that the equipment produced sound of consistent value for all the six measurements taken 1.5 cm from the drilling machine. Equivalent sound level of the room environment measured was 55 dB and that of Portable Rotary Drill Machine was 76 dB.

It may be argued that sound produced from the Portable Rotary Drill Machine itself may affect the sound level measurement during rock drilling. It is important to mention here that if the sound level difference between two sources is more than 10 dB, then the total sound level will remain the same as that of the higher source (Hosrud 2001). Further, taking the measurement very close to the source will reduce the effect of sound produced from other sources.

6.6 SELECTION OF DRILL PARAMETERS

The drilling parameters were fixed after a number of trial runs of the drill machine. Drilling was carried out at 280, 290 and 300 rpm for four different thrusts i.e. 15, 18, 20 and 25 kg/cm². Four different diameter of drill bits were selected i.e. 6 mm, 8 mm, 10 mm and 16 mm for trial run, for shank lengths of 150 mm and 200 mm.

During drilling it was found that the drill bits of diameters lesser than 16 mm was turning to red hot and was unable to withstand the heat. Hence, the minimum diameter of drill rod selected for the present study was 16 mm. In addition to this two more drill bits of diameters 18 mm and 20 mm are also considered. This minimum diameter of drill rod was determined to ensure more number of drill holes on each rock sample/block. Among two shanks i.e. 150 mm and 200 mm lengths, it was concluded to select 200 mm shank in the present study, as it reduce the time required for one complete cycle of drilling. Further, this length was more suitable for the drilling machine as well as for the size of rock blocks selected.

Among three drill speeds i.e. 280, 290 and 300 rpm there was not much variation in the sound levels and the measured values were found to be almost consistent. Hence, in the present study 300 rpm was kept constant for all the trials i.e. different combination of drill parameters for different types of rocks.

6.7 EXPERIMENTAL PROCEDURE

A well prepared rock sample was placed over the wooden base of the Portable Rotary Drill Machine and firmly tightened with a pair of sample holders. For each rock block six vertical holes were drilled by keeping drill bit speed constant at 300 rpm. This was repeated for all three drill rods of diameters 16 mm, 18 mm and 20 mm by varying thrust at four stages i.e. 15, 18, 20 and 25 kg/cm². The sound levels were recorded precisely using dosimeter during drilling. For each rock sample six readings were recorded for one combination of drill diameter and thrust. Once the drill bits were found to be worn out, it was replaced with a new one, so as to ensure consistent in drilling. This procedure was repeated for all the rock samples/rock blocks which are under investigation.

The sound level data captured by the dosimeter for each rock sample was transferred to the computer system and was analysed using Blaze Software. The equivalent sound level of each hole measured/recorded by the dosimeter for different rock types is given in Tables 6.2 to 6.13.

The arithmetic average of each set of six measurements was computed to yield the average A-weighted equivalent sound level for a particular rock block and tabulated with respect to different physico mechanical properties. It was found that the recorded equivalent sound levels were almost consistent.

6.8 RESULTS AND DISCUSSION

6.8.1 Experimental Values With 16 mm Drill Bit and 15 Kg/cm² Thrust

Table-6.2: Equivalent sound levels (Leq) with 16 mm diameter drill rod at 15 kg/cm² thrust

SI. No.	Material	S1(dB)	S2(dB)	S3(dB)	S4(dB)	S5(db)	S6(dB)	S _{AVG} (dB)
1	Granite Grey	96.0	95.2	95.8	95.6	95.7	95.2	95.58
2	AptiteAnantapur	98.5	97.9	98.6	98.3	98.7	98.6	98.43
3	Felsite Mysore	100.1	100.1	100.2	100.1	100.0	100.1	100.1
4	Gabbro Greenish.	101.6	101.5	101.3	101.1	101.1	101.3	101.3
5	Granite Pink Mysore	101.9	101.8	101.7	101.6	101.6	101.7	101.7
6	Syenite	102.0	102.2	102.0	102.3	102.3	102.0	102.1
7	Granite Porphyry	105.6	105.5	105.6	105.6	105.6	105.5	105.5
8	Basalt Nagpur	109.1	109.2	109.2	109.2	109.1	109.0	109.1
9	Syenite Porphyry	110.8	110.9	110.9	110.6	110.6	110.9	110.7
10	Diorite Porphyry	113.3	113.2	113.1	113.3	113.0	113.1	113.1
11	Granite Karnataka	116.5	116.5	116.2	116.2	116.1	116.2	116.2
12	Gabbro Madduru	118.8	118.9	118.7	118.7	118.6	118.9	118.7

From the above table 6.2 for the drill bit parameters of 16 mm dia meter and thrust of 15 kg/cm²it was observed that Granite Grey has a least A weighted sound level of 95.58 dB(A), where as the maximum average A weighted sound level measurement of 118.7 dB(A) was recorded for igneous rock block namely Gabbro Madduru, thereby measuring a difference of 23.12 dB(A) between the rock block of maximum Uni – axial compressive strength (UCS) of 102.6 MPa and the minimum UCS of 46.23 MPa. Whereas the Tensile Strength was found to be 5.23 MPa and 12.3 MPa for respective igneous rock blocks.

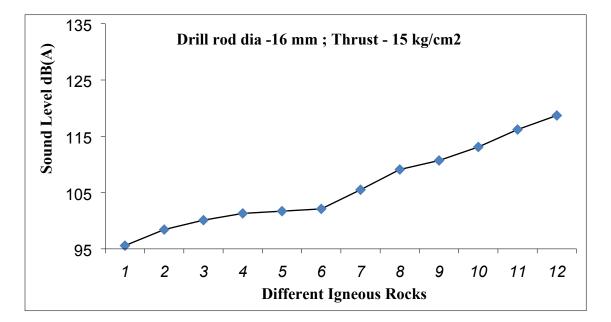


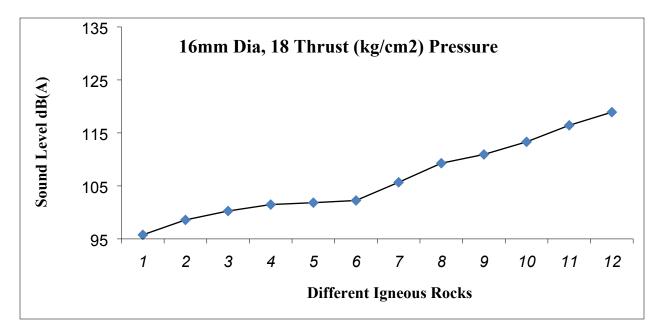
Fig. 6.3 Sound Level Vs rock type for 16 mm diameter drill rod at 15 kg/cm² thrust

6.8.2 Experimental Values With 16 mm Drill Bit and 18 Kg/cm² Thrust

SI. No.	Material	S1(dB)	S2(dB)	S3(dB)	S4(dB)	S5(db)	S6(dB)	S _{AVG} (dB)
1	Granite Grey	96.1	95.3	95.9	95.9	95.8	95.6	95.76667
2	AptiteAnantapur	98.7	98.2	98.7	98.4	98.8	98.7	98.58333
3	Felsite Mysore	100.3	100.3	100.3	100.3	100.1	100.3	100.2667
4	Gabbro Greenish.	101.7	101.6	101.5	101.3	101.3	101.5	101.4833
5	Granite Pink Mysore	102.1	101.9	101.8	101.7	101.7	101.8	101.8333
6	Syenite	102.2	102.3	102.1	102.4	102.3	102.2	102.25
7	Granite Porphyry	105.7	105.6	105.8	105.7	105.7	105.6	105.6833
8	Basalt Nagpur	109.3	109.3	109.3	109.3	109.3	109.2	109.2833
9	Syenite Porphyry	110.9	111.1	111.1	110.7	110.7	111.1	110.9333
10	Diorite Porphyry	113.5	113.3	113.2	113.4	113.3	113.2	113.3167
11	Granite Karnataka	116.7	116.7	116.3	116.3	116.3	116.3	116.4333
12	Gabbro Madduru	118.9	118.7	118.8	118.9	119.1	119.1	118.9167

Table-6.3: Equivalent sound levels (Leq) with 16 mm diameter drill rod at 18 kg/cm² thrust

From the above table 6.3 for the drill bit parameters of 16 mm dia meter and thrust of 18 kg/cm²it was observed that Granite Grey has a least A weighted sound level of 95.76 dB(A), where as the maximum average A weighted sound level measurement of 118.91 dB(A) was recorded for igneous rock block namely Gabbro Madduru, thereby measuring a difference of 23.15 dB(A) between the rock block of maximum Uni – axial compressive strength (UCS) of 102.6 MPa and the minimum UCS of 46.23 MPa. Whereas the Tensile Strength was found to be 5.23 MPa and 12.3 MPa for respective igneous rock blocks.





6.8.3 Experimental Values With 16 mm Drill Bit and 20 Kg/cm ² Thrust

SI. No.	Material	S1(dB)	S2(dB)	S3(dB)	S4(dB)	S5(db)	S6(dB)	S _{AVG} (dB)
1	Granite Grey	97.3	97.3	97.3	97.2	97.4	97.2	97.28
2	AptiteAnantapur	99.1	98.7	98.3	98.8	98.7	98.9	98.75
3	Felsite Mysore	100.5	101.6	100.8	100.8	100.7	100.8	100.86
4	Gabbro Greenish.	102.1	101.9	101.9	101.6	101.7	101.6	101.8
5	Granite Pink Mysore	102.5	102.4	102.5	102.6	102.7	102.4	102.51
6	Syenite	102.6	102.6	102.7	102.7	102.8	102.7	102.68
7	Granite Porphyry	105.9	105.9	105.9	105.9	106.2	106.3	106.01
8	Basalt Nagpur	109.7	109.9	109.9	110.0	110.3	109.8	109.93
9	Syenite Porphyry	111.3	111.4	111.5	111.4	111.6	111.6	111.46
10	Diorite Porphyry	113.6	113.7	113.7	113.8	113.7	113.9	113.73
11	Granite Karnataka	117.0	117.1	116.9	117.1	116.9	116.9	116.98
12	Gabbro Madduru	119.3	119.4	119.4	119.3	119.2	119.3	119.31

Table-6.4: Equivalent sound levels (Leq) with 16 mm diameter drill rod at 20
kg/cm² thrust

From the above table 6.4 for the drill bit parameters of 16 mm dia meter and thrust of 20 kg/cm²it was observed that Granite Grey has a least A weighted sound level of 97.28

dB(A), where as the maximum average A weighted sound level measurement of 119.31 dB(A) was recorded for igneous rock block namely Gabbro Madduru, thereby measuring a difference of 22.03 dB(A) between the rock block of maximum Uni – axial compressive strength (UCS) of 102.6 MPa and the minimum UCS of 46.23 MPa. Whereas the Tensile Strength was found to be 5.23 MPa and 12.3 MPa for respective igneous rock blocks.

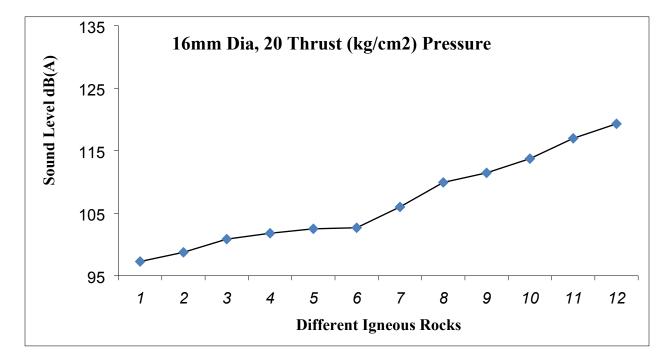
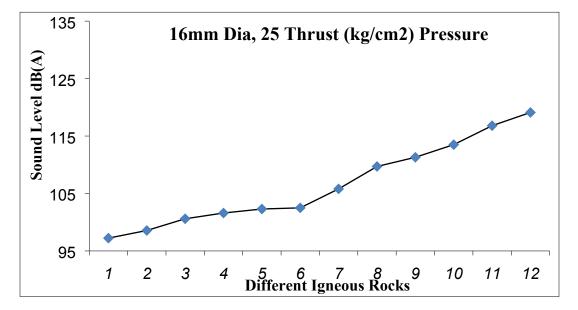


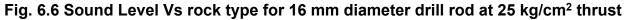
Fig. 6.5 Sound Level Vs rock type for 16 mm diameter drill rod at 20 kg/cm² thrust

6.8.4 Experimental Values With 16 mm Drill Bit and 25 Kg/cm² Thrust

SI. No.	Material	S1(dB)	S2(dB)	S3(dB)	S4(dB)	S5(db)	S6(dB)	S _{AVG} (dB)
1	Granite Grey	97.0	97.1	96.9	97.2	97.3	97.9	97.23
2	AptiteAnantapur	98.9	98.5	98.0	98.7	98.6	98.69	98.56
3	Felsite Mysore	100.2	101.5	100.6	100.6	100.5	100.6	100.6
4	Gabbro Greenish.	101.9	101.8	101.7	101.5	101.6	101.5	101.6
5	Granite Pink Mysore	102.3	102.3	102.3	102.5	102.5	102.2	102.3
6	Syenite	102.5	102.5	102.6	102.6	102.6	102.5	102.5
7	Granite Porphyry	105.8	105.7	105.7	105.8	105.9	106.0	105.8
8	Basalt Nagpur	109.6	109.8	109.8	109.9	109.9	109.7	109.7
9	Syenite Porphyry	111.0	111.2	111.3	111.3	111.5	111.5	111.3
10	Diorite Porphyry	113.5	113.6	113.6	113.6	113.5	113.7	113.5
11	Granite Karnataka	116.9	116.9	116.8	116.9	116.8	116.7	116.8
12	Gabbro Madduru	119.1	119.2	119.2	119.0	119.1	119.1	119.1

Table-6.5: Equivalent sound levels (Leq) with 16 mm diameter drill rod at 25kg/cm² thrust





From the above table 6.5 for the drill bit parameters of 16 mm dia meter and thrust of 25 kg/cm²it was observed that Granite Grey has a least A weighted sound level of 97.23 dB(A), where as the maximum average A weighted sound level measurement of 119.1 dB(A) was recorded for igneous rock block namely Gabbro Madduru, thereby measuring a difference of 21.87 dB(A) between the rock block of maximum Uni – axial compressive strength (UCS) of 102.6 MPa and the minimum UCS of 46.23 MPa. Whereas the Tensile Strength was found to be 5.23 MPa and 12.3 MPa for respective igneous rock blocks.

6.8.5 Experimental Values With 18 mm Drill Bit and 15 Kg/cm² Thrust

SI. No.	Material	S1(dB)	S2(dB)	S3(dB)	S4(dB)	S5(db)	S6(dB)	S _{AVG} (dB)
1	Granite Grey	96.9	96.3	96.1	96.2	96.3	96.0	96.3
2	AptiteAnantapur	98.8	98.9	98.7	98.3	98.7	99.3	98.78
3	Felsite Mysore	100.5	100.6	100.7	100.6	100.5	100.7	100.6
4	Gabbro Greenish.	101.8	101.7	101.7	101.7	101.7	101.9	101.6
5	Granite Pink Mysore	102.3	102.2	102.0	102.0	102.2	102.2	102.1
6	Syenite	102.5	102.6	102.6	102.6	102.7	102.5	102.5
7	Granite Porphyry	105.9	106.0	106.0	105.9	105.8	105.7	105.8
8	Basalt Nagpur	109.3	109.5	109.3	109.6	109.6	109.3	109.4
9	Syenite Porphyry	111.0	111.3	111.3	111.0	111.2	111.2	111.1
10	Diorite Porphyry	113.5	113.6	113.5	113.6	113.6	113.6	113.5
11	Granite Karnataka	116.8	116.9	116.9	116.8	116.9	116.8	116.8
12	Gabbro Madduru	119.0	119.1	119.1	119.1	119.2	119.0	119.0

Table-6.6: Equivalent sound levels (Leq) with 18 mm diameter drill rod at 15 kg/cm² thrust

From the above table 6.6 for the drill bit parameters of 18 mm dia meter and thrust of 15 kg/cm²it was observed that Granite Grey has a least A weighted sound level of 96.3

dB(A), where as the maximum average A weighted sound level measurement of 119.0 dB(A) was recorded for igneous rock block namely Gabbro Madduru, thereby measuring a difference of 22.7 dB(A) between the rock block of maximum Uni – axial compressive strength (UCS) of 102.6 MPa and the minimum UCS of 46.23 MPa. Whereas the Tensile Strength was found to be 5.23 MPa and 12.3 MPa for respective igneous rock blocks.

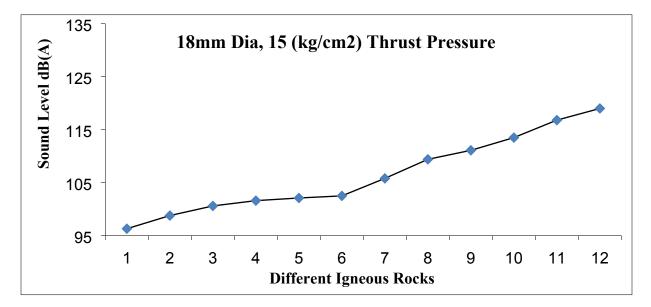


Fig. 6.7 Sound Level Vs rock type for 16 mm diameter drill rod at 15 kg/cm² thrust

6.8.6 Experimental Values With 18 mm Drill Bit and 18 Kg/cm² Thrust

Table-6.7: Equivalent sound levels (Leq) with 18 mm diameter drill rod at 18 kg/cm² thrust

SI. No.	Material	S1(dB)	S2(dB)	S3(dB)	S4(dB)	S5(db)	S6(dB)	S _{AVG} (dB)
1	Granite Grey	97.1	96.4	96.4	96.3	96.4	96.2	96.46
2	AptiteAnantapur	98.8	99.2	98.8	98.5	98.8	99.5	98.93
3	Felsite Mysore	100.7	100.8	100.9	100.7	100.7	100.9	100.78
4	Gabbro Greenish.	101.9	101.8	101.8	101.8	101.8	101.9	101.83
5	Granite Pink Mysore	102.5	102.3	102.3	102.1	102.3	102.4	102.3
6	Syenite	102.7	102.7	102.7	102.7	102.8	102.7	102.7
7	Granite Porphyry	106.2	106.3	106.3	106.2	105.9	105.8	105.45

8	Basalt Nagpur	109.5	109.6	109.4	109.8	109.8	109.4	109.58
9	Syenite Porphyry	111.2	111.5	111.4	111.3	111.3	111.3	111.33
10	Diorite Porphyry	113.6	113.7	113.6	113.7	113.7	113.8	113.68
11	Granite Karnataka	116.9	117.1	117.1	116.9	117.1	116.9	116.5
12	Gabbro Madduru	119.3	119.3	119.2	119.3	119.2	119.2	119.25

From the above table 6.7 for the drill bit parameters of 18 mm dia meter and thrust of 18 kg/cm²it was observed that Granite Grey has a least A weighted sound level of 96.46 dB(A), where as the maximum average A weighted sound level measurement of 119.25 dB(A) was recorded for igneous rock block namely Gabbro Madduru, thereby measuring a difference of 22.79 dB(A) between the rock block of maximum Uni – axial compressive strength (UCS) of 102.6 MPa and the minimum UCS of 46.23 MPa. Whereas the Tensile Strength was found to be 5.23 MPa and 12.3 MPa for respective igneous rock blocks.

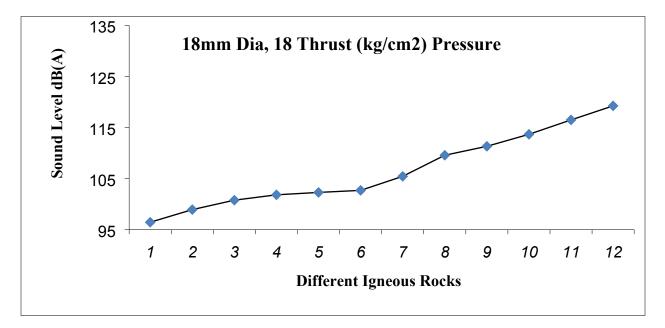


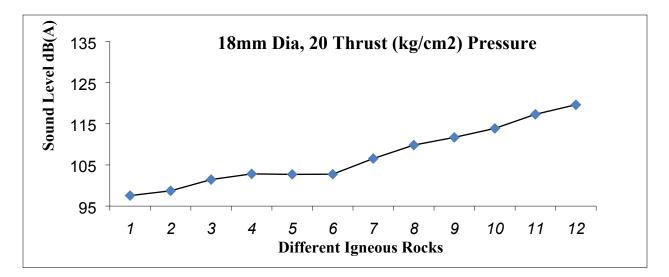
Fig. 6.8 Sound Level Vs rock type for 18 mm diameter drill rod at 18 kg/cm² thrust

6.8.7 Experimental Values With 18 mm Drill Bit and 20 Kg/cm² Thrust

SI. No.	Material	S1(dB)	S2(dB)	S3(dB)	S4(dB)	S5(db)	S6(dB)	S _{AVG} (dB)
1	Granite Grey	97.3	97.8	97.8	97.6	97.7	97.1	97.55
2	AptiteAnantapur	99.0	99.1	98.5	98.4	98.8	98.5	98.71
3	Felsite Mysore	101.3	101.5	101.3	101.3	101.6	101.7	101.45
4	Gabbro Greenish.	102.1	102.4	102.6	102.5	102.5	102.2	102.83
5	Granite Pink Mysore	102.7	102.8	102.6	102.7	102.7	102.8	102.71
6	Syenite	102.9	102.9	102.9	102.8	102.1	102.9	102.75
7	Granite Porphyry	106.3	106.5	106.4	106.5	106.7	106.9	106.55
8	Basalt Nagpur	110.1	109.9	109.8	110.0	109.7	109.8	109.83
9	Syenite Porphyry	111.8	111.7	111.6	111.6	111.7	111.8	111.7
10	Diorite Porphyry	113.9	113.8	113.9	113.9	113.9	113.9	113.88
11	Granite Karnataka	117.1	117.3	117.3	117.4	117.5	117.3	117.31
12	Gabbro Madduru	119.7	119.7	119.7	119.5	119.7	119.5	119.63

Table-6.8: Equivalent sound levels (Leq) with 18 mm diameter drill rod at 20 kg/cm² thrust

From the above table 6.8 for the drill bit parameters of 18 mm dia meter and thrust of 20 kg/cm²it was observed that Granite Grey has a least A weighted sound level of 97.55 dB(A), where as the maximum average A weighted sound level measurement of 119.63 dB(A) was recorded for igneous rock block namely Gabbro Madduru, thereby measuring a difference of 22.08 dB(A) between the rock block of maximum Uni – axial compressive strength (UCS) of 102.6 MPa and the minimum UCS of 46.23 MPa. Whereas the Tensile Strength was found to be 5.23 MPa and 12.3 MPa for respective igneous rock blocks.



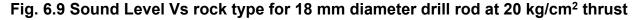


Table-6.9: Equivalent sound levels (Leq) with 18 mm diameter drill rod at 25
kg/cm² thrust

SI. No.	Material	S1(dB)	S2(dB)	S3(dB)	S4(dB)	S5(db)	S6(dB)	S _{AVG} (dB)
1	Granite Grey	97.2	97.6	97.5	97.3	97.5	97.0	97.35
2	AptiteAnantapur	98.9	98.9	98.2	98.1	98.7	98.3	98.5
3	Felsite Mysore	101.0	101.3	101.1	101.1	101.5	101.5	101.25
4	Gabbro Greenish.	102.0	102.2	102.3	102.3	102.3	102.0	102.1
5	Granite Pink Mysore	102.6	102.7	102.5	102.5	102.5	102.6	102.5
6	Syenite	102.8	102.7	102.7	102.7	102.8	102.8	102.7
7	Granite Porphyry	106.2	106.3	106.3	106.3	106.5	106.7	106.3
8	Basalt Nagpur	109.9	109.8	109.7	109.9	109.6	109.6	109.7
9	Syenite Porphyry	111.6	111.5	111.5	111.5	111.6	111.7	111.5
10	Diorite Porphyry	113.8	113.7	113.7	113.7	113.8	113.8	113.7
11	Granite Karnataka	117.0	117.2	117.1	117.2	117.3	117.2	117.1
12	Gabbro Madduru	119.5	119.6	119.5	119.3	119.6	119.3	119.4

From the above table 6.9 for the drill bit parameters of 18 mm dia meter and thrust of 25 kg/cm²it was observed that Granite Grey has a least A weighted sound level of 97.35 dB(A), where as the maximum average A weighted sound level measurement of 119.40 dB(A) was recorded for igneous rock block namely Gabbro Madduru, thereby

measuring a difference of 22.05 dB(A) between the rock block of maximum Uni – axial compressive strength (UCS) of 102.6 MPa and the minimum UCS of 46.23 MPa. Whereas the Tensile Strength was found to be 5.23 MPa and 12.3 MPa for respective igneous rock blocks.

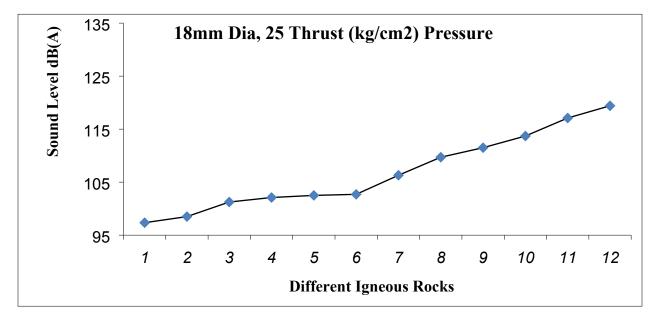


Fig. 6.10 Sound Level Vs rock type for 18 mm diameter drill rod at 25 kg/cm² thrust

6.8.9 Experimental Values With 20 mm Drill Bit and 15 Kg/cm² Thrust

SI. No.	Material	S1(dB)	S2(dB)	S3(dB)	S4(dB)	S5(db)	S6(dB)	S _{AVG} (dB)
1	Granite Grey	96.8	96.1	97.0	97.2	97.3	96.9	96.88
2	AptiteAnantapur	98.9	98.3	99.5	99.6	99.1	99.2	99.1
3	Felsite Mysore	101.0	100.9	101.0	100.9	101.1	101.3	101
4	Gabbro Greenish.	102.0	102.0	102.3	102.3	102.3	102.0	102.1
5	Granite Pink Mysore	102.3	102.5	102.5	102.6	102.6	102.5	102.5
6	Syenite	102.7	102.8	102.8	102.7	102.8	102.7	102.7
7	Granite Porphyry	106.2	106.1	106.6	106.6	106.5	106.5	106.4
8	Basalt Nagpur	109.8	109.7	109.7	109.8	109.8	109.7	109.7
9	Syenite Porphyry	111.5	111.5	111.6	111.5	111.5	111.6	111.5

Table-6.10: Equivalent sound levels (Leq) with 20 mm diameter drill rod at 15kg/cm² thrust

10	Diorite Porphyry	113.7	113.7	113.8	113.8	113.8	113.7	113.7
11	Granite Karnataka	117.0	117.1	117.2	117.1	117.2	117.0	117.1
12	Gabbro Madduru	119.5	119.3	119.5	119.5	119.3	119.5	119.4

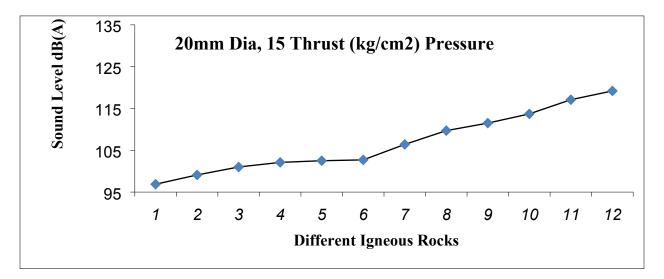


Fig. 6.11 Sound Level Vs rock type for 20 mm diameter drill rod at 15 kg/cm² thrust

From the above table 6.10 for the drill bit parameters of 20 mm dia meter and thrust of 15 kg/cm²it was observed that Granite Grey has a least A weighted sound level of 96.88 dB(A), where as the maximum average A weighted sound level measurement of 119.40 dB(A) was recorded for igneous rock block namely Gabbro Madduru, thereby measuring a difference of 22.52 dB(A) between the rock block of maximum Uni – axial compressive strength (UCS) of 102.6 MPa and the minimum UCS of 46.23 MPa. Whereas the Tensile Strength was found to be 5.23 MPa and 12.3 MPa for respective igneous rock blocks.

6.8.10 Experimental Values With 20 mm Drill Bit and 18 Kg/cm² Thrust

SI. No.	Material	S1(dB)	S2(dB)	S3(dB)	S4(dB)	S5(db)	S6(dB)	S _{AVG} (dB)
1	Granite Grey	96.9	96.4	96.3	97.1	97.5	97.1	96.88
2	AptiteAnantapur	99.2	98.5	99.6	99.7	99.3	99.3	99.3
3	Felsite Mysore	101.3	101.1	101.3	101.1	101.3	101.5	101.26
4	Gabbro Greenish.	102.2	102.2	102.5	102.5	102.4	102.3	102.35
5	Granite Pink Mysore	102.5	102.6	102.6	102.7	102.8	102.6	102.63
6	Syenite	102.9	102.9	102.9	102.8	102.9	102.9	102.88
7	Granite Porphyry	106.3	106.3	106.7	106.8	106.6	106.7	106.56
8	Basalt Nagpur	109.9	109.8	109.9	109.9	109.9	109.8	109.86
9	Syenite Porphyry	111.7	111.6	111.7	111.6	111.7	111.7	111.66
10	Diorite Porphyry	113.8	113.8	113.9	113.9	113.9	113.9	113.86
11	Granite Karnataka	117.1	117.3	117.3	117.3	117.3	117.2	117.25
12	Gabbro Madduru	119.7	119.5	119.7	119.7	119.6	119.7	119.65

Table-6.11: Equivalent sound levels (Leq) with 20 mm diameter drill rod at 18kg/cm² thrust

From the above table 6.11 for the drill bit parameters of 20 mm dia meter and thrust of 18 kg/cm²it was observed that Granite Grey has a least A weighted sound level of 96.88 dB(A), where as the maximum average A weighted sound level measurement of 119.65 dB(A) was recorded for igneous rock block namely Gabbro Madduru, thereby measuring a difference of 22.77 dB(A) between the rock block of maximum Uni – axial compressive strength (UCS) of 102.6 MPa and the minimum UCS of 46.23 MPa. Whereas the Tensile Strength was found to be 5.23 MPa and 12.3 MPa for respective igneous rock blocks.

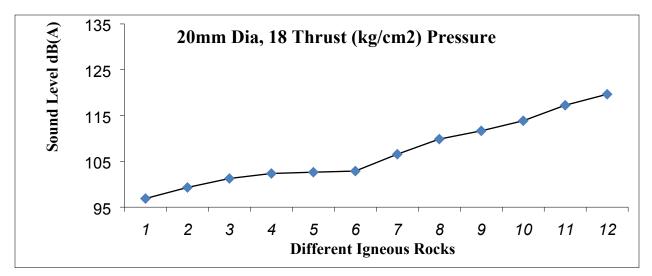


Fig. 6.12 Sound Level Vs rock type for 20 mm diameter drill rod at 18 kg/cm² thrust

6.8.11 Experimental	l Values With 20 n	nm Drill Bit and 20 Kg/cm ²	Thrust
---------------------	--------------------	--	--------

Table-6.12: Equivalent sound levels (Leq) with 20 mm diameter drill rod at 20
kg/cm² thrust

SI. No.	Material	S1(dB)	S2(dB)	S3(dB)	S4(dB)	S5(db)	S6(dB)	S _{AVG} (dB)
1	Granite Grey	97.9	96.3	96.3	97.8	98.1	97.2	97.26
2	AptiteAnantapur	99.7	99.8	99.8	99.5	99.7	99.7	99.7
3	Felsite Mysore	101.9	101.8	101.9	102.1	101.9	101.8	101.9
4	Gabbro Greenish.	102.6	103.1	102.9	102.3	102.5	102.7	102.68
5	Granite Pink Mysore	103.1	103.1	102.9	103.1	102.9	103.0	103.01
6	Syenite	103.1	103.0	102.9	103.1	103.1	102.9	103.01
7	Granite Porphyry	106.9	107.1	107.0	107.2	106.8	106.9	106.98
8	Basalt Nagpur	110.1	110.2	110.1	110.3	110.3	110.5	110.25
9	Syenite Porphyry	111.9	112.1	112.1	112.1	111.6	111.9	111.95
10	Diorite Porphyry	114.0	114.1	113.9	113.9	114.0	114.1	114
11	Granite Karnataka	117.6	117.7	117.7	117.8	117.7	117.6	117.68
12	Gabbro Madduru	119.9	119.8	119.9	120.0	119.9	119.9	119.9

From the above table 6.12 for the drill bit parameters of 20 mm dia meter and thrust of 20 kg/cm²it was observed that Granite Grey has a least A weighted sound level of 97.26 dB(A), whereas the maximum average A weighted sound level measurement of 119.9

dB(A) was recorded for igneous rock block namely Gabbro Madduru, thereby measuring a difference of 22.64 dB(A) between the rock block of maximum Uni – axial compressive strength (UCS) of 102.6 MPa and the minimum UCS of 46.23 MPa. Whereas the Tensile Strength was found to be 5.23 MPa and 12.3 MPa for respective igneous rock blocks.

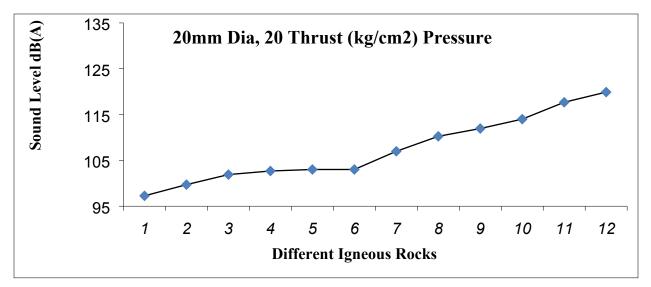


Fig. 6.13 Sound Level Vs rock type for 20 mm diameter drill rod at 20 kg/cm² thrust

6.8.12 Experimental Values With 20 mm Drill Bit and 25 Kg/cm² Thrust

SI. No.	Material	S1(dB)	S2(dB)	S3(dB)	S4(dB)	S5(db)	S6(dB)	S _{AVG} (dB)
1	Granite Grey	97.8	96.2	97.6	97.8	97.9	97.0	97.38
2	AptiteAnantapur	99.5	99.7	99.7	99.3	99.6	99.5	99.55
3	Felsite Mysore	101.8	101.7	101.7	101.9	101.8	101.6	101.75
4	Gabbro Greenish.	102.5	102.9	102.8	102.2	102.3	102.5	102.53
5	Granite Pink Mysore	102.9	102.9	102.8	102.9	102.7	102.9	102.8
6	Syenite	102.9	102.9	102.8	102.9	102.9	102.8	102.8
7	Granite Porphyry	106.8	106.9	106.9	106.9	106.7	106.7	106.8
8	Basalt Nagpur	110.0	110.1	110.0	110.1	110.1	110.3	110.1
9	Syenite Porphyry	111.8	111.9	111.9	111.9	111.8	111.7	111.8

Table-6.13: Equivalent sound levels (Leq) with 20 mm diameter drill rod at 25
kg/cm² thrust

10	Diorite Porphyry	113.9	113.9	113.8	113.8	113.9	113.9	113.8
11	Granite Karnataka	117.5	117.5	117.6	117.6	117.6	117.5	117.5
12	Gabbro Madduru	119.8	119.7	119.8	119.9	119.8	119.8	119.8

From the above table 6.13 for the drill bit parameters of 20 mm dia meter and thrust of 25 kg/cm²it was observed that Granite Grey has a least A weighted sound level of 97.38 dB(A), where as the maximum average A weighted sound level measurement of 119.8 dB(A) was recorded for igneous rock block namely Gabbro Madduru, thereby measuring a difference of 22.42 dB(A) between the rock block of maximum Uni – axial compressive strength (UCS) of 102.6 MPa and the minimum UCS of 46.23 MPa. Whereas the Tensile Strength was found to be 5.23 MPa and 12.3 MPa for respective igneous rock blocks.

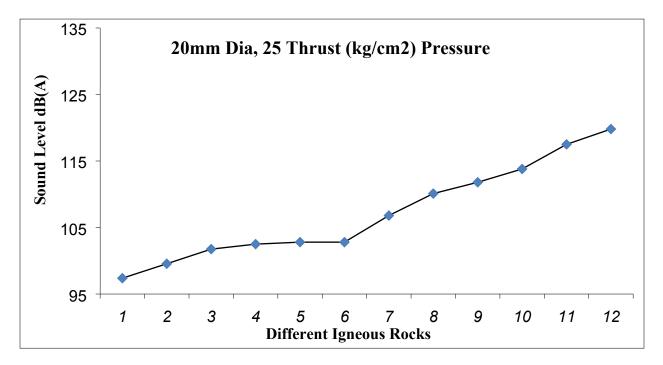


Fig. 6.14 Sound Level Vs rock type for 20 mm diameter drill rod at 25 kg/cm² thrust

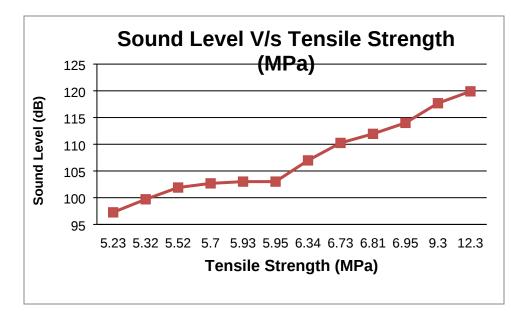


Fig 6.15 : Sound Level measurement using Portable Drill Setup by considering drill bit diameter 18 (mm) and Thrust Pressure 20(kg/cm²) V/s Tensile Strength.

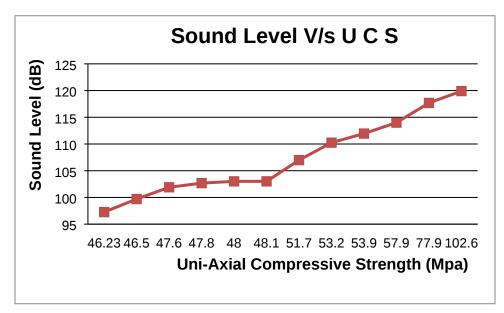


Fig 6.16: Sound Level measurement using Portable Drill Setup by considering drill bit diameter 18 (mm) and Thrust Pressure 20(kg/cm²) V/s UCS.

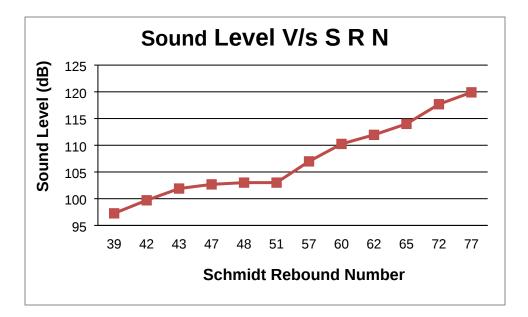


Fig 6.17: Sound Level measurement using Portable Drill Setup by considering drill bit diameter 18 (mm) and Thrust Pressure 20(kg/cm²) V/s SRN.

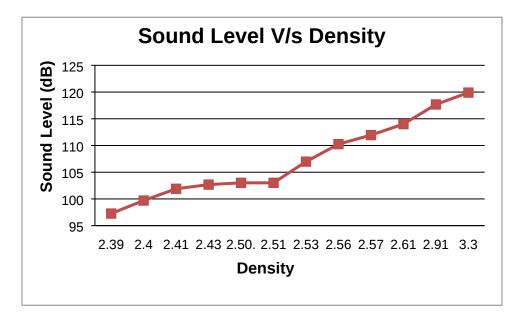


Fig 6.18: Sound Level measurement using Portable Drill Setup by considering drill bit diameter 18 (mm) and Thrust Pressure 20(kg/cm²) V/s Density.

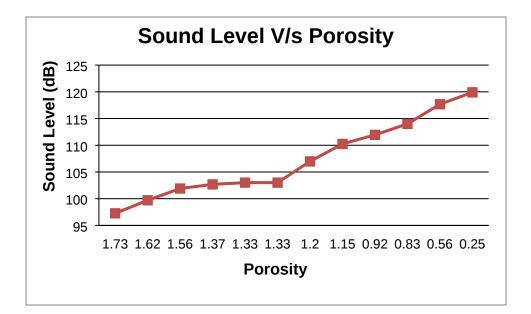


Fig 6.19: Sound Level measurement using Portable Drill Setup by considering drill bit diameter 18 (mm) and Thrust Pressure 20(kg/cm²) V/s Porosity.

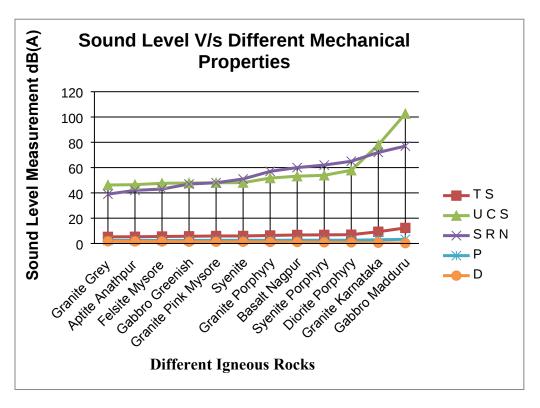


Fig 6.20: Sound Level measurement using Portable Drill Setup by considering drill bit diameter 18 (mm) and Thrust Pressure 20(kg/cm²) V/s Different Mechanical Properties.

6.9 INFERENCE

From the figures shown above, it can be observed that A-weighted equivalent sound level produced during drilling process increases non linearly as the mechanical properties like UCS, SRN, Density, porosity and Tensile strength of the igneous rock increases. This may be due to increase in resistance offered against drilling. Further It may be argued that sound produced from the fabricated drill set up itself may affect the sound level measurement during rock drilling. It is important to mention here that the motor used in the set-up is noiseless with negligible sound level and hence do not have any impact on the equivalent sound level measurements.

CHAPTER 7

MATHEMATICAL MODELLING

7.1 INTRODUCTION

In statistics, regression analysis is a statistical process for the estimation and the determination of relationship among variables (Finol et al). It consists many techniques for modeling and analyzing several variables, when the main aim is on the relationship between a dependent variable and one or more independent variables. The regression analysis helps to understand how the typical value of the dependent variable (or 'criterion variable') changes when any one of the independent variable is varied, while the other independent variables are kept constant. Most commonly, regression analysis determines the conditional expectation of the dependent variable when the independent variables are known. In all instances, the estimation target is a function of the independent variables called the regression function. In regression analysis, it is also possible to characterize the variation of the dependent variable around the regression function which can be described by a probability distribution.

Most common application of regression analysis is for prediction and forecasting. It is also used to know which among the independent variables are related to the dependent variable, and determination of different forms of these relationships. There are many techniques for carrying out regression analysis, among them linear regression and ordinary least squares regression are significant, where regression function is defined in terms of a finite number of unknown parameters, that are estimated from the available data. In practice, the performance of regression

analysis methods depends on the form of the data generating process, and how it depends on the regression approach being followed. Since the actual form of the data-generating process is generally not known, regression analysis often depends to some extent on making assumptions about this process. These assumptions are sometimes testable if a sufficient quantity of data is known. Regression models for prediction are often applicable even when the assumptions are moderately violated, although they may not perform optimally. However, in certain applications with small effects or questions of causality based on observational data, regression methods can give misleading results.

7.2 UNKNOWN PARAMETERS

In various fields of engineering application, different terminologies are used in place of dependent and independent variables.

- The unknown parameters, denoted as β, which may represent a scalar or a vector.
- The independent variables X.
- The dependent variable Y.

A regression model relates 'Y' to a function of 'X' and ' β ' can be represented as shown in equation 7.1

To carry out regression analysis, the form of the function 'f' must be specified. Sometimes the form of this function is based on knowledge about the relationship between Y and X that does not rely on the data. If no such knowledge is available, a flexible or convenient form for f is chosen. In order to perform a regression analysis it is necessary to provide information about the dependent variable Y.

• If *N* is the number of measurements and *K* being the number of unknown parameters, if *N*datais of the form (*Y*, *X*), where N < K, most classical approaches to regression analysis cannot be performed, since the system of equations defining the regression model is underdetermined, as sufficient data is unavailable to recover β .

• If exactly N = K data points are observed, and the function *f* is linear, the equations $Y = f(X, \beta)$ can be solved exactly rather than approximately. This reduces to solving a set of *N* equations with *N* unknowns (the elements of β), which has a unique solution as long as the X are linearly independent. If *f* is nonlinear, a solution may not exist, or many solutions may exist.

• The most common situation is where N > K data points are observed. In this case, there is enough information in the data to estimate a unique value for β that best fits the data in some sense, and the regression model when applied to the data can be viewed as an over determined system in β .

In the last case, the regression analysis provides the tools for:

1. Finding a solution for unknown parameters β that will, for example, minimize difference between the measured and predicted values of the dependent variable Y (also known as method of least squares).

2. Under certain statistical assumptions, the regression analysis may use the surplus information to provide statistical information about the unknown parameters β and predicted values of the dependent variable Y.

7.3 NECESSARY NUMBER OF INDEPENDENT MEASUREMENTS

If three unknown parameters are considered in a regression model (such as β_0 , β_1 , and β_2)and, if any investigator performs ten measurements all at exactly the same value of independent variable vector X (which contains the independent variables X_1 , X_2 , and X_3), the regression analysis will not be applicable to give a unique set of estimated values for the three unknown parameters. In that case, it is most appropriate to estimate the average value and the standard deviation of the dependent variable Y. Similarly, measuring at two different values of X can provide enough data for a regression with two unknowns, but not for three or more unknowns.

If the investigator had performed measurements at three different values of the independent variable vector X, then regression analysis would provide a unique set of estimates for the three unknown parameters in β .

7.3.1 Statistical Assumptions

When the number of measurements 'N' is larger than the number of unknown parameters 'K' and the measurement errors are normally distributed then the excess of information contained in (N - K) measurements is applicable to make statistical predictions about the unknown parameters. This excess of information is referred to as the degrees of freedom of the regression.

7.4 UNDERLYING ASSUMPTIONS

Classical assumptions for regression analysis include:

• The sample is representative of the population for the inference prediction.

- The error is a random variable with a mean of zero conditional on the explanatory variables.
- The independent variables are measured with no error. Instead, modeling may be done using errors-in-variables model techniques if any errors do exist.
- The predictors are linearly independent, i.e. it is not possible to express any predictor as a linear combination of the others.
- The errors are uncorrelated, that is, the variance–covariance matrix of the errors is diagonal and each non-zero element is the variance of the error.

These above mentioned conditions are sufficient for the least-squares estimator to possess desirable properties; in particular, these assumptions imply that the parameter estimates will be unbiased, consistent, and efficient in the class of linear unbiased estimators. It is important to note that actual data less commonly satisfies the assumptions. That is, the method is useful even though the assumptions are not correct. Variation from the assumptions can sometimes be used as a measure of how far the model is from being applicable. Most of the assumptions may be relaxed in more advanced methods. Reports of statistical analyses usually include analyses of tests on the sample data and methodology for the fit and usefulness of the model.

Independent and dependent variables often refer to values measured at point locations. There may be spatial trends and spatial autocorrelation in the variables that violate statistical assumptions of regression. Geographic weighted regression is one technique to deal with such data. Also, variables may include values aggregated by areas. With aggregated data the modifiable areal unit problem can cause extreme variation in regression parameters. In linear regression, the model specification is

that the dependent variable, y_i is a linear combination of the parameters (but need not be linear in the independent variables). For a particular instance in simple linear regression for modeling n data points there is one independent variable: x_i and two parameters, β_0 and β_1 , which is given by the equation for a straight line as shown in equation 7.2

Straight line: $y_i = \beta_0 + \beta_1 x_i + \varepsilon_i$, i=1... n. 7.2

7.5 MATHEMATICAL MODELS USING MULTIPLE REGRESSION ANALYSIS

As discussed earlier, multiple regression analysis is a powerful technique used for predicting the unknown value of a variable from the known value of two or more variables- also called the predictors. The variable whose value is to be predicted is known as the dependent variable and the ones whose known values are used for prediction are known independent (exploratory) variables.

In general, the multiple regression equation of Y on $X_1, X_2... X_k$ can be of the form as shown in equation 7.3

$$Y = b_0 + b_1 X_1 + b_2 X_2 + \dots + b_k X_k \qquad \dots 7.3$$

Here b_0 is the intercept and b_1 , b_2 , b_3 ... b_k are analogous to the slope in linear regression equation and are also called regression coefficients. They can be interpreted the same way as slope. Thus if $b_i = 2.5$, it would indicates that Y will increase by 2.5 units if X_i increased by 1 unit. The appropriateness of the multiple regression model as a whole can be tested by the F-test in the ANOVA table. A significant F indicates a linear relationship between Y and at least one of the X's. Once a multiple regression equation has been constructed, one can check how good it is (in terms of predictive ability) by examining the coefficient of determination (R2).

R2 always lies between 0 and 1.All software provides it whenever regression procedure is run. The closer R_2 is to 1, the better is the model and its prediction.

A related question is whether the independent variables individually influence the dependent variable significantly. Statistically, it is equivalent to testing the null hypothesis that the relevant regression coefficient is zero. This can be done using t-test. If the t-test of a regression coefficient is significant, it indicates that the variable is in question influences Y significantly, while controlling for other independent explanatory variables. Multiple regression technique does not test whether data are linear. On the contrary, it proceeds by assuming that the relationship between the Y and each of X_i's is linear. Hence as a rule, it is prudent to always look at the scatter plots of (Y, X_i), i= 1, 2... k. If any plot suggests non linearity, one may use a suitable transformation to attain linearity.

Another important assumption is nonexistence of multi co linearity, the independent variables are not related among themselves. At a very basic level, this can be tested by computing the correlation coefficient between each pair of independent variables. Therefore, multiple regression analysis is used when one is interested in predicting a continuous dependent variable from a number of independent variables. If dependent variable is dichotomous, then logistic regression should be used.

Further it can be expected score on one variable R, R Square and Adjusted R Square, where R is a measure of the correlation between the observed value and the predicted value of the criterion variable. R Square (R²) is the square of this measure of correlation and indicates the proportion of the variance in the criterion variable, so an Adjusted R Square value is calculated which takes into account the number of variables in the model and the number of observations the model is based on. This Adjusted R Square value gives the most useful measure of the success of

the model. If, for example we have an Adjusted R Square value of 0.75 we can say that the model is accounted for 75% of the variance in the criterion.

7.6 INPUT TO A REGRESSION PROBLEM

Simple regression: (x1, Y1), (x1, Y2)... (x n, Y n)

Multiple regression: ((x1)1, (x2)1, (x3)1 ... (x K) 1, Y1),

((x1)2, (x2)2, (x3) 2 ... (x K)2, Y2),

((x1)3, (x2)3, (x3)3, ... (x K)3, Y3), ... ,

((x1) n, (x2) n, (x3)n, ... (x K)n, Y n),

The variable Y is designated as the "dependent variable." The only distinction between the two situations above is whether there is just one x predictor or many. The predictors are called "independent variables." There is certain awkwardness about giving generic names for the independent variables in the multiple regression case. In this notation, x1 is the name of the first independent variable, and its values are (x1)1, (x1)2, (x1)3... (x1) n.

The listing for the multiple regression case suggests that the data are found in a spreadsheet. In application programs like Minitab, the variables can appear in any of the spreadsheet columns. The dependent variable and the independent variables may appear in any columns in any order.

When choosing a predictor variable selection is based, such that one might be able to correlate with the criterion variable, but that is not strongly correlated with the other predictor variables. However, correlations amongst the predictor variables are not unusual. The term multi co linearity is used to describe the situation when a high correlation is detected between two or more predictor variables. Such high

correlations cause problems when trying to draw inferences about the relative contribution of each predictor variable to the success of the model.

There are different ways that the relative contribution of each predictor variable can be assessed. The success of the model in predicting the criterion variable is then assessed. In contrast, "hierarchical" methods enter the variables into the model in a specified order. The order specified should reflect some theoretical consideration or previous findings. If one have no reason to believe that one variable is likely to be more important than other this method is not applicable. As each variable is entered into the model its contribution is assessed. If adding the variable does not significantly increase the predictive power of the model then the variable is dropped.

Further, in this research work pertaining to rock properties, to obtain applicable and practical predictive qualitative relationships, it is necessary to model the rock properties and the drill process variables. Multiple regression analysis is widely used for modeling and analyzing the experimental results. Therefore in this present investigation, Regression is the determination of statistical relationship between two or more variables. It is the method to deal with the formulation of mathematical model depicting the relationship among the variables which can be used for the purpose of prediction of values of dependent or response variables, given the values of predictor or independent variable(s), the analysis concerning the relationship is known as multiple correlation and equations describing such relationships are called as multiple regression modeling, the statistical methodology used to relate variables (Bowerman and O'connell, 1990).

In order to establish the predictive models among the parameters obtained in this research work, multiple regression and Analysis of Variance (ANOVA) techniques

are used. For modeling and analysis Minitab 15 software for windows was used. Further, higher p-value means the independent variables are not statistically significant and it can also be stated if the p-value is set at 0.05 then to reject the null hypothesis, the significance level of 5.001 % or higher has to be used. Hence the pvalue represents the lowest significance level that can be used to reject the null hypothesis.

The p-values for individual predictor variables will be obtain from t –statistics and the same value for individual variables assess the significance of including that predictor in the regression analysis. On the other part p-value obtain from the F – statistics assess, whether the whole regression is significant or not. If one predictor is significant, so will be the regression and for the overall regression the value obtained at R-Squared and R-adjusted has to be considered.

7.7 MULTIPLE REGRESSION ANALYSIS AND ANOVA TECHNIQUE

The laboratory experimental results are used to model the various responses using multiple regression method by using a non-linear fit among the responses and the corresponding significant parameters. The performance of the model depends on a large number of factors that act and interact in a complex manner. When the predictor variables in a multiple regression model are interrelated or are dependent on each other, a multi-co linearity problem exists and hinders the ability to assess the importance of a predictor variable. The solutions to the problem are to remove one or more of the highly correlated predictor variables or to add more scenarios used in building the model.

The mathematical modeling of sound level produced during drilling is influenced by many factors. Therefore a detailed process representation anticipates a second

order model. ANOVA is carried out to find which input parameter significantly affects the desired response. To facilitate the experiments and measurement, four important predictor variables are considered in the present study. They are drill bit diameter (A), drill bit speed (B), Thrust applied (C) and the average value of A – weighted equivalent sound level produced during drilling in dB(A) i.e., (D). The responses considered are Uni-axial compressive strength (UCS), Schmidt rebound number (SRN), Dry density (r), Tensile strength (TS), and Percentage porosity (p). The mathematical models for the physico-mechanical properties with parameters under consideration can be of the form as shown in equation 7.4

$$Y = \Box (x_1, x_2, x_3 ...) + \Box 7.4$$

where, Y is the response and x_1 , x_2 , x_3 are the independent process variables and \mathcal{E} is fitting error. A quadratic model of fcan be written as shown in equation 7.5

$$\Box = b_0 + \sum_{i=1}^n b_i x_i + \sum_{i=1}^n b_{ij} x_i^2 + \sum_{i$$

where, b_i represents the linear effect of x_i , b_{ij} represents the quadratic effect of x_i and b_{ij} in fourth term represents linear interaction between x_i and x_j . Then the regression models contain linear terms, squared terms and cross product terms.

Individual responses are modeled by using multiple regression analysis. The coefficient table lists the estimated coefficients for all the predictors. p-value determines the observed relationship between response and the predictors and indicates, whether it is statistically significant or not. If the p-value is less than the selected a level (to test the significance, one needs to set a risk level called the alpha level. In most cases, the 'rule of thumb' is to set the alpha level at 0.05, i.e., 95% confidence interval), the association is statistically significant and the model is selected. On the other hand, if it is more than the selected *a* level, it is not statistically

significant and the independent variable under consideration is removed from the model and the regression analysis is performed by using all the remaining independent variables. The procedure is continued by removing one independent variable at a time from the model. The screening is stopped when the independent variable remaining in the model can not be removed from the system.

The criterion for the explanation capability of the dependent variable from information obtained via the independent variables is the strength of the relation (R^2). When the explained variation is divided by the total variation of *y* values, i.e. when R^2 is calculated, then the proportion of variation of the dependent variable changes is obtained and that is explained by the independent variable. As the number of independent variables increases, R^2 will be greater. Therefore, R^2 should be adjusted (corrected) as follows, which is shown in equation 7.6

$$\overline{R}^{2} = \left[R^{2} - \frac{k-1}{n-1} \right] \left[\frac{n-1}{n-k} \right] \qquad \dots 7.6$$

where, *k* the number of model parameters, and *n* the number of scenarios. Higher the value of R^2 means that there is a good correlation between the experimentally measured values and the predicted values using the developed models.

7.7.1Selection of Rock Samples for Modeling

For experimental works, different categories of rocks are collected. Out of the total collected rocks, there are 12 igneous rock types. Out of these, 10 rock types are used for developing the model and 2 types are used to test the accuracy of the developed model. Therefore as can be seen by considering the drilling parameters for set of three different conditions as stated below:

- a) Drill diameters of 16, 18 and 20 mm.
- b) Drill speed of the machine 280,290 and 300 rpm.
- c) Thrust pressure of 15, 18, 20, and 25kg/cm²

Hence, the total conditions used for developing the mathematical model are 3 (drill bits) x 3 (machine speed) x 4 (applied thrust pressure) i.e. in total 36x10 (360 Leq values) and similarly 36x2 (72 Leg values) for testing the developed models.

7.8 MATHEMATICAL REGRESSION MODEL FOR DIFFERENT IGNEOUS ROCK PROPERTIES:

Multiple regression model to predict uni-axial compressive strength for igneous rock types is given by equation 7.7:

UCS (Igneous) = 525.9816+2.135 x A+0.065 x B+2.5 x C -10.162 x D + 0.061 x D² – 0.026 x A x D -0.001 x B x D- 0.02811 x C x D7.7

Significance of regression coefficients for estimation of uni-axial compressive strength is listed in Table 7.1, which also shows *t* value and p-value. The final ANOVA table of the reduced quadratic model for UCS is shown in Table 7.2. This table also represents degrees of freedom (DF), mean square (MS), sum of squares (SS), *F*-value and p-value associated with factors. As seen from Table 7.3, for igneous rocks, the selected model explains 95.1% of the total variation in the observed UCS tests.

Model terms for UCS	Parameter estimate (coefficients)	t-value	p-value
Constant	525.9816	21.693	0.000
А	2.135	7.9	0.000
В	0.065	2.3	0.006
С	2.5	1.9	0.042
D	-10.162	-26.365	0.000
D ²	0.061	32.117	0.000
AD	-0.026	-9.2	0.000
BD	-0.001	-3.12	0.002
CD	-0.02811	-2.293	0.021

TABLE 7.1: Significance of regression coefficients for estimation of uniaxial compressive strength (igneous rocks)

TABLE 7.2: Analysis of variance (ANOVA) for the selected quadratic model for estimation of UCS (igneous rocks)

Source of variations	Degree of freedom	Sum of squares	Mean squares	F-value	p-value
Model	8	216597.60	26623.07	2956.60	0.000
Linear	4	6625.92	1639.73	175.39	0.000
Square	1	9726.65	9712.64	1023.75	0.000
Interaction	3	932.55	321.18	32.65	0.000
Residual Error	631	5216.60	9.03	-	-
Total	639	22501.72	-	-	-

TABLE 7.3: Model summary for dependent variable (UCS - igneous rocks)

R ²	Predicted R ²	Adjusted R ²	Standard error
0.963	0.96	0.9512	2.98

Multiple regression models to predict Schmidt rebound number of the igneous rock types is given by equation 7.8:

SRN (Igneous) =11.7894+ 0.582 x A + 0.021 x B - 2.357 x D + 0.018 x D² - 0.008 x

· ..7.8

AxD

Significance of regression coefficients for estimation of Schmidt rebound number is listed in Table 7.4, which also shows *t* value and p-value. The final ANOVA table of the reduced quadratic model for SRN is shown in Table 7.5. This table also represents degrees of freedom (DF), mean square (MS), sum of squares (SS), F-value and p-value associated with factors. As seen from Table 7.6, for igneous rocks, the selected model explains 96.8% of the total variation in the observed SRN tests.

TABLE 7.4: Significance of regression coefficients for estimation of Schmidtrebound number (igneous rocks)

Model terms for SRN	Parameter estimate (coefficients)	t-value	p-value
Constant	11.7894	14.013	0.000
A	0.582	5.136	0.000
В	0.021	1.675	0.046
D	-2.357	-15.357	0.000
D ²	0.018	21.562	0.000
AD	-0.008	-6.612	0.000

TABLE 7.5: Analysis of variance (ANOVA) for the selected quadratic model for estimation of Schmidt rebound number (igneous rocks)

Source of variations	Degree of freedom	Sum of squares	Mean squares	F-value	p- value
Model	5	55238.82	9210.15	5474.98	0.000
Linear	3	415.46	142.12	84.32	0.000
Square	1	823.28	848.300	503.18	0.000
Interaction	1	82.65	41.615	24.68	0.000
Residual Error	628	1065.16	1.686	-	-
Total	633	2386.55	-	-	-

TABLE 7.6: Model summary for dependent variable (SRN - igneous rocks)

R ²	Predicted R ²	Adjusted R ²	Standard error	
0.9612	0.9702	0.9682	1.28	

Multiple regression model to predict dry density of the igneous rock types is given by equation 7.9:

ρ Density (Igneous) = 1.5 + 0.387 x A -0.1813 x D + 0.0010 x D² – 0.00039 x A X D

.....7.9

Significance of regression coefficients for estimation of dry density is listed in Table 7.7, which also shows t-value and p-value. The final ANOVA table of the reduced quadratic model for dry density is shown in Table 7.8. This table also represents degrees of freedom (DF), mean square (MS), sum of squares (SS), F-value and p-value associated with factors. As seen from Table 7.9, for igneous rocks, the selected model explains 75.6% of the total variation in the observed dry density tests.

TABLE 7.7: Significance of regression coefficients for estimation of Dry
density (igneous rock)

Model terms for Dry density	Parameter estimate (coefficients)	t-value	p-value
Constant	1.5	19.120	0.000
A	0.387	4.814	0.000
D	-0.1813	-16.284	0.000
D ²	0.0010	17.966	0.000
AD	-0.00039	-5.205	0.000

TABLE 7.8: Analysis of variance (ANOVA) for the selected quadratic model for estimation of Dry density (igneous rock)

Source of variations	Degree of freedom	Sum of squares	Mean squares	F-value	p-value
Model	4	16.52150	4.23376	552.05	0.000
Linear	2	2.36246	1.03123	134.47	0.000
Square	1	2.57550	2.47550	322.79	0.000
Interaction	1	0.20780	0.20780	27.10	0.000
Residual Error	635	4.56991	0.00767	-	-
Total	639	9.71567	-	-	-

TABLE 7.9: Model summary for dependent variable (Dry Density – igneous rock)

R ²	Predicted R ²	Adjusted R ²	Standard error	
0.756	0.752	0.7563	0.0865	

Multiple regression models to predict Tensile strength of the igneous rock types is given by equation 7.10:

TS (Igneous) =56.518714 +0.2730 x A .0086 x B + 0.3106 x C -1.2657 x D + .0076 x

D² - .0032 x A x D – 0.0001 x B x D -0.0035 x C x D7.10

Significance of regression coefficients for estimation of Tensile strength is listed in Table 7.10, which also shows t-value and p-value. The final ANOVA table of the reduced quadratic model for Tensile strength is shown in Table 7.11. This table also represents degrees of freedom (DF), mean square (MS), sum of squares (SS), F-value and p-value associated with factors. As seen from Table 7.12, for igneous rocks, the selected model explains 95.6% of the total variation in the observed Tensile strength tests.

Model terms for Tensile strength	Parameter estimate (coefficients)	t-value	
Constant	56.418714	22.778	0.000
А	0.2730	8.219	0.000
В	0.0086	2.768	0.006
С	0.3106	2.012	0.045
D	-1.2657	-27.448	0.000
D ²	0.0076	34.392	0.000
AD	-0.0031	-9.783	0.000
BD	-0.0001	-3.295	0.001
CD	-0.0035	-2.394	0.017

 TABLE 7.10: Significance of regression coefficients for estimation of Tensile

 strength (igneous rock)

TABLE 7.11: Analysis of variance (ANOVA) for the selected quadratic model for estimation of Tensile strength (igneous rock)

Source of variations	Degree of freedom	Sum of squares	Mean squares	F-value	p- value
Model	8	3855.630	481.954	3691.64	0.000
Linear	4	104.024	26.006	199.20	0.000
Square	1	154.417	154.417	1182.80	0.000
Interaction	3	14.491	4.830	37.00	0.000
Residual Error	628	82.379	0.131	-	-
Total	636	3938.01	-	-	-

TABLE 7.12: Model summary for dependent variable (Tensile strength - igneous rock).

R ²	Predicted R ²	Adjusted R ²	Standard error
0.961	0.9823	0.9562	0.362

Multiple regression model to predict percentage porosity of the igneous rock types is given by equation 7.11:

P (Igneous) = 32.151974+0.0943 x A – 0.5338 x D + .00023 x D² – 0.0007 x A x D.

.....7.11

Significance of regression coefficients for estimation of percentage porosity is listed in Table 7.13, which also shows t-value and p-value. The final ANOVA table of the reduced quadratic model for dry percentage porosity is shown in Table 7.14. This table also represents degrees of freedom (DF), mean square (MS), sum of squares (SS), F-value and p-value associated with factors. As seen from Table 7.15, for igneous rocks, the selected model explains 93.59% of the total variation in the observed percentage porosity tests.

TABLE 7.13: Significance of regression coefficients for estimation of %
Porosity (igneous rock)

Model terms for % Porosity	Parameter estimate (coefficients)	t-value	p-value
Constant	32.151974	26.815	0.000
A	0.0943	5.493	0.000
D	-0.5338	-22.467	0.000
D ²	0.00023	18.923	0.000
AD	-0.0007	-4.513	0.000

TABLE 7.14: Analysis of variance (ANOVA) for the selected quadratic model for estimation of % Porosity (igneous rock)

Source of variations	Degree of freedom	Sum of squares	Mean squares	F-value	p-value
Model	4	361.8989	90.4747	2591.90	0.000
Linear	2	17.6990	8.8495	253.52	0.000
Square	1	12.4997	12.4997	358.09	0.000
Interaction	1	0.7109	0.7109	20.37	0.000
Residual Error	635	22.1657	0.0349	-	-
Total	639	384.065	-	-	-

TABLE 7.15: Model summary for dependent variable (% Porosity - igneous
rock)

R ²	Predicted R ²	Adjusted R ²	Standard error
0.9313	0.9215	0.9359	0.186

7.9 PERFORMANCE PREDICTION OF THE DERIVED MODELS:

In fact, the coefficient of correlation between the measured and predicted values is a good indicator to check the prediction performance of the model. However, in this study, Values Account For (VAF) (Equation) and Root Mean Square Error (RMSE) (Equation) indices were calculated to compare the performance of the prediction capacity of predictive models developed (Alvarez and Babuska 1999, Finol et al. 2001, Gokceoglu 2002, Yilmaz and Yuksek 2008, Yilmaz and Yuksek 2009, Yilmaz and Kaynar 2011) .Further, the VAF value would be of the form as shown equation 7.12 and RMSE would be of the form as shown in equation 7.13

$$VAF = \left[1 - \frac{var(y - y')}{var(y)}\right] \times 100 \qquad \dots 7.12$$

$$RMSE = \sqrt{\frac{1}{N}\sum_{i=1}^{N} (y - y')^{2}} \qquad \dots 7.13$$

Where y and y' are the measured and predicted values respectively. If the VAF is 100 and RMSE is 0, then the model will be excellent. Mean absolute percentage error (MAPE) which is a measure of accuracy in a fitted series value was also used to check the prediction performances of the models. MAPE usually expresses accuracy as a percentage as shown in below equation 7.14

$$MAPE = \frac{1}{N} \sum_{i=1}^{N} \left| \frac{A_i \cdot P_i}{A_i} \right| \times 100 \qquad \dots 7.14$$

Where A_i is the actual value and P_i is the predicted value. Lower values of MAPE, indicate that there will be a better correlation between predicted values and experimental results.

Using the developed regression models for igneous rocks, performance prediction indices for training as well as test data were calculated and are given in Table 7.16. From the table it is evident that the developed model for predicting dry density is less efficient when compared to all other models as it has low VAF value. MAPE values for test data are 12.3, 16.1, 21.2, 8.3, and 7.21 for UCS, SRN, dry density, TS, and percentage porosity respectively, which indirectly explains the reliability of the predicted models of igneous rocks.

TABLE 7.16 Performance prediction indices of the regression models (IgneousRock)

		UCS	SRN	Density	Tensile strength	% Porosity
D.	VAF	95.3	96.1	75.6	93.192	
raining data	RMSE	5.5	6.1	0.39	0.38	0.506
	MAPE	10.2	12.5	15.2	5.21	6.601
st dê	VAF	93.6	95.3	62.3	95.7	92.23
Test	RMSE	8.2	7.96	0.62	0.625	0.523

	MAPE	12.3	16.1	21.2	8.3	7.21
7.10 INFEREN	CE:					

In this chapter, the experimental values obtained in the present investigation are used to develop mathematical models for different phsico-mechanical properties, such as Uni-axial compressive strength, Tensile strength, SRN, Density and porosity. By using the Minitab software and Analysis of variance, models are established. It is observed that physico-mechanical properties has very low p-values with acceptable tolerance and hence the performance of prediction indices were determined which further can be used to compare with the developed models for other indirect investigation methods.

CHAPTER - 8

CONCLUSION AND SCOPE FOR FURTHER RESEARCH

8.1 CONCLUSIONS

The following conclusions are drawn based on the research work carried out:

- The drilling set-up designed and fabricated in this research investigation will be useful in the estimation of rock properties using sound level produced during drilling.
- The results of laboratory studies carried out in this research work clearly indicate the prediction of the physico-mechanical properties of igneous rocks with acceptable degree of accuracy.
- Since the proposed method is considerably simple and requires lesser time in arriving at the physico-mechanical properties, hence it will be an economically feasible (in traditional method, the rock blocks are sent from the mine to some established laboratory which is time consuming as well as a costly affair) and a better alternative and could find wide application in various geo-technical field.
- In order to establish the predictive models among different parameters considered in the investigation, it is found that multiple regression analysis can be an effective technique in minimizing the uncertainties involved in the investigation.

- The set of empirical equations developed in this research work for prediction of physico-mechanical properties of igneous rocks can be readily utilized in mining and various geotechnical fields.
- Though, an attempt has been made in this research work to estimate rock properties using sound level produced during drilling using the various empirical equations developed, however, it is not aimed at replacing the suggested ISRM methods. The developed equations can be certainly used for a quick and easy estimate of different rock properties using the light weighted, portable, low cost drill set-up with acceptable degree of accuracy.
- Further, the proposed equations can serve a valuable information for the purpose of comparing the performance of the developed models with other indirect investigations.

8.2 SCOPE FOR FURTHER RESEARCH

- In the present work only igneous rock samples collected across the southern part of India has been used. However this investigation can be extended for sedimentary as well as metamorphic rock samples.
- In the present work, drill bit diameters of 16 mm, 18 mm and 20 mm have been used for the purpose of drilling. This work can be extended for other drill diameters too.
- Also, in the present work only intact rocks are considered for recording the sound level produced during the process of drilling. Similar work can be extended/carried out for the rock mass drilling.

 The fabricated portable drill set-up can be directly tried in the field and prediction models could be developed and the results of the same could be compared with that of IRSM suggested methods.

REFERENCES

Aleotti, L., Poletto, F., Miranda, F., Corubolo, P., Abramo, F. & Craglietto, A. (1999); "Seismic while drilling technology: use and analysis of the drill-bit seismic source in a cross-hole survey";Geophys Prospect;47:25–39.

Aleotti, L., Poletto, F., Miranda, F., Corubolo, P., Abramo, F. and Craglietto, A. (1999). "Seismic while drilling technology: use and analysis of the drill-bit seismic source in a cross-hole survey." *Geophys. Prospect.*, 47, 25–39.

Altindag, R. (2002). "The evaluation of rock brittleness concept on rotary blast hole drills." *J. South Afr. Ins. Min. and Metall.*, January/February 2002, 61-66.

Alvarez, G.M. and Babuska, R. (1999). "Fuzzy model for the prediction of unconfined compressive strength of rock samples." *Int. J. Rock Mech. Min. Sci.*, 36, 339-349.

Alvarez, G.M., Bruines, P.A. and Verhoef, P.N.W. (2000). "Modeling tunnel boring machine performance by neuro-fuzzy methods." *Tunnelling Underground Space Technol.*, 15(3), 259-269.

Anderson, J.A. (1995). An introduction to neural networks. The MIT Press.

Asanuma, H. & Niitsuma H. (1996); "Triaxial seismic measurement while drilling and estimation of subsurface structure"; Int J Rock Mech Min SciGeomech Abstr; 33(7): 307A.

Asanuma, H. and Niitsuma, H. (1996). "Triaxial seismic measurement while drilling and estimation of subsurface structure." *Int. J. Rock Mech. Min. Sci. Geomech. Abstr.*, 33(7), 307A.

Aufmuth, R.E. (1973). "A systematic determination of engineering criteria for rocks." *Bull. Int. Assoc. Eng. Geol.*, 11, 235-245.

Aydin, A. (2009). "ISRM suggested method for determination of schmidt hammer rebound hardness: revised version." *Int. J. Rock Mech. Min. Sci,*. 46, 627-634.

Bai, J., Wild, S., Ware, J.A. and Sabir, B.B. (2003). "Using neural networks to predict workability of concrete incorporating metakaolin and fly ash." *Adv. Eng. Softw.*, 34(11-12), 663–669.

Basma, A.A. and Kallas, N. (2004). "Modeling soil collapse by artificial neural networks." *Geotech. Geol. Eng.*, 22, 427-438.

Benardos, A.G. and Kaliampakos, D.C. (2004). "Modelling TBM performance with artificial neural networks." *Tunnelling Underground Space Technol.*, 19, 597-605.

Beverly, B.E., Schoenwolf, D.A. and Brierly, G.S. (1979). "Correlations of rock index values with engineering properties and the classification of intact rock." FHWA, Washington, D.C.

Bilgin, N., Eskikaya, S. and Dincer, T. (1993). "The performance analysis of large diameter blast hole rotary drills." *J. Min. Mech. Autom.*, 129-135.

Bowerman, B.L. and O'connell, R.T. (1990). *Linear Statistical Models*, Duxbury Press, p. 1024.

Bradford, I.D.R., Fuller, J., Thompson, P.J. and Walsgrove, T.R. (1998). "Benefits of assessing the solids production risk in a North Sea reservoir using elastoplastic modeling." *Proc., SPE/ISRM Eurock '98,* Trondheim, Norway, 261-269.

Byerlee, J. (1978). "A review of rock mechanics studies in the United States pertinent to earthquake prediction." *Pure Appl. Geophys.*, 116, 586-602.

Cargill, J.S. and Shakoor, A. (1990). "Evaluation of empirical methods for measuring the uniaxial compressive strength." *Int. J. Rock. Mech. Min. Sci.*, 27, 495-503.

Carmichael, R.S. (1982). *Handbook of physical properties of rocks – Vol. II*. CRC Press, Boca Raton.

Chang, C., Zoback, M.D. and Khaksar, A. (2006). "Empirical relations between rock strength and physical properties in sedimentary rocks." *J. Pet. Sci. Eng.*, 51(3-4), 223-237.

Colin, H. (2005). *Noise control from concept to application*. Taylor and Francis, New York.

Das, S.K. and Basudhar, P. (2008). "Prediction of residual friction angle of clay artificial neural network." *Eng. Geol.*, 100, 142-145.

Deere, D.U. and Miller, R.P. (1966). "Engineering classification and index properties for intact rocks." *Tech. Report. Air Force Weapons Lab.*, New Mexico, No., AFNL-TR, Kirtland, 65-116.

Delibalta, M. S., Kahraman, S. &Comakli, R. (2015); "The usability of noise level from rock cutting for the prediction of physico-mechanical properties of rocks"; Fluctuation and Noise Levels – An Interdisciplinary Scientific Journal on Random Processes in Physical, Biological and Technological Systems; Vol. 14(1); 12 pages.

Faisal, I.S., Edward, J.C. and Omar, H.A. (2007). "Estimation of rock engineering properties using hardness tests", *Eng. Geol.*, 90, 138-147.

Fernandez, J.V. & Pixton, D.S. (2005);"Integrated drilling system using mud actuated down hole hammer as primary engine"; Final Technical Report 34365R05, DOE Award Number: DE-FC26-97FT34365. NOVATEK Provo, Utah.

Finol, J., Guo, Y.K. and Jing, X.D. (2001). "A rule based fuzzy model for the prediction of petrophysical rock parameters." *J. Pet. Sci. Eng.*, 29, 97-113.

Fletcher, P., Coveney, P.V., Hughes, T.L. and Methven, C.M. (1995). "Predicting quality and performance of oilfield cements with artificial neural networks and FTIR spectroscopy. *J. Pet. Technol.*, 47(2), 129-130.

Futo, J., Usalova, L. and Ivanicova, L. (2003). "Optimization of rock disintegration using the acoustic signal." *J. Acta Montanistica Slovaca*, 8(4), 174-175.

Ghose, A.K. and Chakraborti, S. (1986). "Empirical strength indices of Indian coals an investigation." *Proc., 27th US Symp. on Rock Mech.*, Balkema, Rotterdam, 59-61.

Gill, D.E., Corthesy, R. and Leite, M.H. (2005). "Determining the minimal number of specimens for laboratory testing of rock properties." *J. Eng. Geol.*, 78(1-2), 29-51.

Gokceoglu, C. (2002). "A fuzzy triangular chart to predict the uniaxial compressive strength of the Ankara Agglomerates from their petrographic composition." *Eng. Geol.*, 66, 39-51.

Gradl, C., Eustes, A.W. and Thonhauser, G. (2008). "An Analysis of Noise Characteristics of Drill Bits." *Society of Petrolium Engineers (SPE), Annual Technical Conference*, Denver, Colorado, USA.

Gunaydin, O. (2009). "Estimation of compaction parameters by using statistical analyses and artificial neural networks." *Environ. Geol.*, 57, 203-215.

Hadjahmadi, A.H., Homayounpour, M.M. and Ahadi, S.M. (2008). "Robust Weighted Fuzzy C-Means Clustering." *Proc., IEEE Int. Conf. on Fuzzy Systems* (*FUZZ2008*), 305-311.

Hand, M., Rueter, C., Evans, B.J., Dodds, K. & Addis, T.(1999);"Look-ahead prediction of pore pressure while drilling: assessment of existing and promising technologies. Doc GRI-99/0042, Gas Res Inst Chicago.

Haramy, K.Y. and DeMarco, M.J. (1985). "Use of Schmidt hammer for rock and coal testing." *Proc.*, *26th US Symp. on Rock Mech.*, Rapid City, 26-28 June, 549–555.

Hardy, H.R. Jr. (1972). "Application of acoustic emission technique to rock mechanics research." *Acoustic emission*, ASTM STP 505, 41-83.

Hardy, H.R. Jr. (1977). "Emergence of acoustic emission, microseismic activity as a tool in geomechanics." *Proc. First Conf. Acoustic Emission/Micro Seismic Activity in Geologic Structures and Materials*, Trans. Tech. Publications. 13-31.

Hatherly, P.J. (2002). "Rock strength assessment from geophysical logging." *Proc., 8th Int. Symp. on Borehole Geophysics for Minerals, Geotechnical and Groundwater Applications*, Toronto, Ontario, 21st-23rd August.

Haykin, S. (1999). *Neural networks – A comprehensive foundation*. Pearson Education Publishing.

Holdo, J. (1958); "Energy consumed by rock drill noise", Mining Magazine.

Horsrud, P. (2001). "Estimating mechanical properties of shale from empirical correlations." *SPE Drill Complet.* 16, 68-73.

Hsu, K. (1997). "Sonic-while-drilling tool detects over pressured formations." *Oil Gas J.*, 59–67.

Hsu, K. (1997); "Sonic-while-drilling tool detects over pressured formations"; Oil Gas J; 59–67.

Hurtado, J.E., Londono, J.M. and Meza, M.A. (2001). "On the applicability of neural networks for soil dynamic amplification analysis." *Soil Dyn. Earthquake Eng.*, 21, 579-591.

Jizba, D. (1991). "Mechanical and acoustical properties of sandstones and shales." PhD thesis, Stanford University.

Jung, S.J., Prisbrey, K. and Wu, G. (1994). "Prediction of rock hardness and drillability using acoustic emission signatures during indentation." *Int. J. Rock Mech. Min. Sci. Geomech. Abstr.*, 31(5), 561-567.

Kahraman S., Korkmazve, S. and Akcay, M. (1996). "The reliability of using Schmidt hammer and point load strength test in assessing uniaxial compressive strength." *K.T.U. Dept. Geol. Eng. 30th year Symp. Book*, Trabzon, 362-369.

Kahraman, S. (2001). "Evaluation of simple methods for assessing the uniaxial compressive strength of rock." *Int. J. Rock Mech. Min. Sci.*, 38(7), 981-994.

Kahraman, S. (2002). "Estimating the direct P-wave velocity value of intact rock from indirect laboratory measurements." *Int. J. Rock Mech. Min. Sci.*, 39, 101-104.

Kahraman, S. and Alber, M. (2006). "Predicting the physico- mechanical properties of rocks from electrical impedance spectroscopy measurements." *Int. J. Rock Mech. Min. Sci.*, 43(4), 543-553.

Kahraman, S., Altun, H., Tezekici, B.S. and Fener, M. (2006). "Sawability prediction of carbonate rocks from shear strength parameters using artificial neural networks." *Int. J. Rock Mech. Min. Sci.*, 43, 157-164.

Kahraman, S., Balci, C., Yazici, S. and Bilgin, N. (2000). "Prediction of the penetration rate of rotary blasthole drills using a new drillability index." *Int. J. Rock Mech. Min. Sci.*, 37(5), 729-743.

Kahraman, S., Gunaydin, O., Alber, M. and Fener, M. (2009). "Evaluating the strength and deformability properties of Misis fault breccia using artificial neural networks." *Expert Syst. Appl.*, 36, 6874-6878.

Kahraman, S., Delibalta, M. S.&Comakli, R. (2013); "Noise level measurement test to predict the abrasion resistance of rock aggregates"; Fluctuation and Noise Levels – An Interdisciplinary Scientific Journal on Random Processes in Physical, Biological and Technological Systems; Vol. 12(4); 12 pages.

Karakurt (2012) "Predictive modelling of noise level generated during sawing of rocks by circular diamond saw blades" Karadeniz Technical University, Trabzon

61080, Turkey. Sadhan a Vol. 38, Part 3, June 2013, pp. 491–511. c Indian Academy of Sciences.

Karakurt, I., Aydin, G. & Aydiner, K. (2013); "Predictive modelling of noise level generated during sawing of rocks by circular diamond sawblades"; Sadhana, Academy Proceedings – Engineering Sciences; Vol.38(3); pp. 491 – 511.

Karakus, M., Kumral, M. and Kilic, O. (2005). "Predicting elastic properties of intact rocks from index tests using multiple regression modelling." *Int. J. Rock Mech. Min. Sci.*, 42,323-330.

Katz, O., Reches, Z. and Roegiers, J.C. (2000). "Evaluation of mechanical rock properties using Schmidt Hammer." *Int. J. Rock Mech. Min. Sci.*, 37(4), 723-728.

Kawasaki, S., Tanimoto, C., Koizumi, K. and Ishikawa, M. (2002). "An attempt to estimate mechanical properties of rocks using the Equotip hardness tester." *J. Jpn. Soc. Eng. Geol.*, 43, 244-248.

Kawasaki, S., Yoshida, M., Tanimoto, C. and Masuya, T. (2000). "The development of property evaluation method for rock materials based on the simple rebound hardness test: investigations on the effects of test conditions and fundamental properties." *J. Jpn. Soc. Eng. Geol.*, 41, 230-241.

Kayadelen, C. (2008). "Estimation of effective stress parameter of unsaturated soils by using artificial neural networks." *Int. J. Numer. Anal. Methods Geomech.*, 32(9), 1087-1106.

Kidybinski, A. (1980). "Bursting liability indices of coal." *Int. J. Rock Mech. Min. Sci. Geomech. Abstr.*, 17, 167-171.

Kilic A. and Teymen A. (2008). "Determination of mechanical properties of rocks using simple Methods." *Bull. Eng. Geol. Environ.*, 67, 237-244.

Kim, K. and Gao, H., (1995). "Probabilistic approaches to estimating variation in the mechanical properties of Rock masses." *Int. J. Rock Mech. Min. Sci. Geomech. Abstr.*, 32(2), 111-120.

Kivade, S. B., Murthy, Ch. S. N. & Vardhan, H. (2014); "Investigation of noise level and penetration rate on pneumatic drill vis-à-vis rock compressive strength and abrasitivity"; Journal of the Institution of Engineers (India): Series D, Metallurgical, Materials and Mining Engineering; Vol. 95(2); pp. 103-114.

Kivade, S. B., Murthy, Ch. S. N. &Vardhan, H. (2011);" Assessment of noise and effect of thrust on penetration rate in percussive drilling"; Journal of the Institution of Engineers (India); Mining Engineering Division; Vol. 91; pp. 3-7.

Kivade, S. B., Murthy, Ch. S. N. &Vardhan, H. (2014); "Laboratory investigation on percussive drilling"; Journal of the Institution of Engineers (India): Series D, Metallurgical, Materials and Mining Engineering; Vol. 94(2); pp. 81-87.

Knill, J.L., Franklin, J.A. and Malone, A.W. (1968). "A study of acoustic emission from stressed rock." *Int. J. Rock Mech. Min. Sci. Geomech. Abstr.*, 16, 23-35.

Kwasniewski, M. (1989). "Rock at great depth." *Laws of brittle failure and of B–D transition in sandstones,* V. Maury and D. Fourmaintraux, eds., Balkema, Brookfield, 45–58.

Lai, S. and Sera, M. (1997). "Concrete strength prediction by mean of neural networks." *Constr. Build. Mater.*, 11(2), 93-98.

Lama, R.D. and Vutukuri, V.S. (1978). *Handbook on mechanical properties of rocks – Vol. II*. Trans. Tech. Publications, Clausthal, Germany.

Lee, I.M. and Lee, J.H. (1996). "Prediction of pile bearing capacity using artificial neural networks." *Comput. Geotech.*, 18(3), 189–200.

Lee, S.J., Lee, S.R. and Kim, Y.S. (2003). "An approach to estimate unsaturated shear strength using artificial neural network and hyperbolic formulation." *Comput. Geotech.*, 30, 489-503.

Levin, R.I. and Rubin, D.S. (2006). "*Statistics for management*," Seventh ed. Prentice Hall of India, New Delhi.

Magali, R.G.M. and Paul E.M.A. (2003). "A comprehensive review for industrial applicability of artificial neural networks." *IEEE Trans. Ind. Electron.*, 50(3), 585-601.

Martinez R. D. (1991);"Formation pressure prediction with seismic data from the Gulf of Mexico. SPE Formation Eval; 6(1):27–32.

McNally, G.H. (1990). "The prediction of geotechnical rock properties from sonic and neutron logs." *Explor. Geophys.*, 21, 65-71.

Miklusova, V., Usalova, L., Ivanicova, L. and Krepelka, F. (2006). "Acoustic signal – new feature in monitoring of rock disintegration process." *Contrib. Geophys. Geodesy*, SAS 36, SI 6SGK, 125-133.

Miller, D.M., Kaminsky, E.J. and Rana, S. (1995). "Neural network classification of remote-sensing data." *Comput. Geosci.*, 21(3), 377-386.

Miller, W. C. (1963); "Noise from pneumatic rock drills—measurement and significance," U.S. Bureau of Mines—Report of Investigation No. 6165, 1–30.

Miranda, F. (1996). "Impact of the seismic while drilling technique on exploration wells." *Int. J. Rock Mech. Min. Sci. Geomech. Abstr.*, 33(8), 360A.

Miranda, F. (1996); "Impact of the seismic while drilling technique on exploration wells"; Int J Rock Mech Min SciGeomechAbstr;33(8):360A.

Mogi, K. (1973). "Rock fracture." Annu. Rev. Earth Planet. Sci., 1, 63-84.

Moulenkamp, F. and Grima, M.A. (1999). "Application of neural networks for the prediction of unconfined compressive strength (UCS) for Equotip hardness." *Int. J. Rock Mech. Min. Sci.*, 36, 29-39.

Neural Network Toolbox ^{™ 6} Mathworks – R2008.

O'Rourke, J.E. (1989). "Rock index properties for geo engineering underground development." *Min. Eng.*, 106-110.

Obert, L. (1941). "Use of sub audible noise for prediction of rock bursts – Part I." *U.S. Bureau of Mines*, R.I. 3555.

Obert, L. and Duvall, W.I. (1942). "Use of sub audible noise for prediction of rock bursts - Part II." *U.S. Bureau of Mines*, R.I. 3654.

Onyia, E.C. (1988); "Relationships between formation strength, drilling strength, and electric log properties" In: SPE 18166, 63rd annual technical conference and exhibition of the SPE. Houston, TX; 2–5 October. p. 605–18.

Powell, W. H. (1956); "The assessment of noise at collieries," Transactions of the Institution of Mining Engineers, 116, 22–42.

Proceq, S.A. (1977a). *Operating instructions concrete test hammer types N and NR*. Zurich, Switzerland.

Proceq, S.A. (1977b). *Equotip operations instructions*. 5th edn. PROCEQ SA Zurich, Switzerland.

Rafiq, M.Y, Bugmann, G. and Easterbrook, D.J. (2001). "Neural network design for engineering applications." *Comput. Struct.*, 79, 1541-1552.

Rajesh Kumar, B., Vardhan, H. & Govindaraj, M. (2010); "Estimating rock properties using sound level during drilling: Field investigation"; International Journal of Mining and Mineral Engineering; Vol.2(3); pp. 169-184.

Rajesh Kumar, B., Vardhan, H. & Govindaraj, M. (2011); "A new approach for estimation of properties of metamorphic rocks"; International Journal of Mining and Mineral Engineering; Vol.3(2); pp. 109-123.

Rajesh Kumar, B., Vardhan, H. & Govindaraj, M. (2011); "Prediction of uniaxial compressive strength, tensile strength and porosity of sedimentary rocks using

sound level produced during rotary drilling"; Rock Mechanics and Rock Engineering; Vol.44(5); pp.613-620.

Rajesh Kumar, B., Vardhan, H. & Govindaraj, M. (2011); "Sound level produced during rock drilling vis-à-vis rock properties"; Engineering Geology; Vol.123(4); pp. 333-337.

Rajesh Kumar, B., Vardhan, H. & Govindaraj, M. (2013); "Artificial neural network model for prediction of rock properties from sound level produced during drilling";Geomechanics and Geoengineering: An International Journal; Vol. 8(1); pp. 53 – 61.

Rajesh Kumar, B., Vardhan, H., Govindaraj, M. & Vijay, G. S. (2013); "Regression analysis and ANN models to predict rock properties from sound levels produced during drilling"; International Journal of Rock Mechanics and Mining Sciences; Vol.58(1); pp. 61-72.

Rasmussen, C.E. (2000). *The Infinite Gaussian Mixture Model. Neural Information Processing Systems*. MIT Press. 12, 554–560.

Rector J. W., Hardage, B. A.(1992);"Radiation pattern and seismic waves generated by a working roller-cone drill bit" Geophysics; 57(10):1319–33.

Rector, J.W. and Hardage, B.A. (1992). "Radiation pattern and seismic waves generated by a working roller-cone drill bit." *Geophysics*, 57(10), 1319–1333.

Romeo, G., Mele, F. and Morelli, A. (1995). "Neural networks and discrimination of seismic signals." *Comput. Geoscie.*, 21(2), 279-288.

Sabatini, P.J., Bachus, R.C., Mayne, P.W., Schneider, J.A. and Zettler, T.E. (2002). "Evaluation of soil and rock properties." *Geotech. Eng.*, Circular No. 5, Report No. FHWA-IF-034, FHWA, US.

Sachapazis, C.I. (1990). "Correlating Schmidt hardness with compressive strength and Young's modulus of carbonate rocks." *Bull. Int. Assoc. Eng. Geol.*, 42,75-83.

Sarkar, K., Tiwary, A. and Singh, T.N. (2010). "Estimation of strength parameters of rock using artificial neural networks." *Bull. Eng. Geol. Environ.* 69(4), 599-606.

Schmidt, E. (1951). "A non-destructive concrete tester." Concrete, 59, 34-35.

Schon, J.H. (1996). "Physical properties of rock – Fundamentals and principles of petrophysics." *Handbook of Geophysical exploration, Seismic exploration*, Pergamon Press, Oxford, p.583

Shamina, O.G. (1975). "Modeling of earthquakes." *Earth Phys.*, 10, 10-20.

Sheorey, P.R., Barat, D., Das, M.N., Mukherjee, K.P. and Singh, B. (1984). "Schmidt hammer rebound data for estimation of large scale in situ coal strength." *Int. J. Rock Mech. Min. Sci.*, 21, 39-42.

Singh, M. and Rao, S.K. (2005). "Empirical methods to estimate the strength of the jointed rock masses." *Eng. Geol.*, 77(1-2), 127-137.

Singh, R.N., Hassani, F.P. and Elkington, P.A.S. (1983). "The application of strength and deformation index testing to the stability assessment of coal measures excavations." *Proc., 24th US Symp. on Rock Mech.*, Texas A & M University, AEG, 599-609.

Singh, V.K., Singh, D. and Singh, T.N. (2001). "Prediction of strength properties of some schistose rocks from petrographic properties using artificial neural networks." *Int. J. Rock Mech. Min. Sci.*, 38, 269-284.

Sinha, S.K. and Wang, M.C. (2007). "Artificial neural network prediction models for soil compaction and permeability." *Geotech. Geol. Eng.*, 26, 47-64.

Sonmez, H., Gokceoglu, C., Nefeslioglu, H.A. and Kayabasi, A. (2006). "Estimation of rock modulus–For intact rocks with an artificial neural network and for rock masses with a new empirical equation." *Int. J. Rock Mech. Min. Sci.*, 43, 224-235. Sonmez, H., Tuncay, E. and Gokceoglu, C. (2004). "Models to predict the uniaxial compressive strength and the modulus of elasticity for Ankara Agglomerate." *Int. J. Rock Mech. Min. Sci.*, 41, 717-729.

SrisharanShreedharan, ChiranthHegde, Sunil Sharma &Vardhan, H. (2014); "Acoustic fingerprinting for rock identification during drilling"; International Journal of Mining and Mineral Engineering; Vol.5(2); pp. 89 - 105.

Stuart, R.K., Charles, F.P. & Hans, T. (2007); "Method for borehole measurement of formation properties"; US patent issued on October 30, 2007 [application No. 10779885 filed on 17–02–2004].

Szlavin, J. (1974). "Relationship between some physical properties of rock determined by laboratory tests." *Int. J. Rock Mech. Min. Sci. Geomech. Abstr.*, 11(2), 57-66.

Szwedzicki, T. (1998). "Technical Note – Indentation hardness testing of rock." *Int. J. Rock Mech. Min. Sci.*, 36(6), 825-829.

Taskiran, T. (2010). "Prediction of California bearing ratio (CBR) of fine grained soils by AI methods." *Adv. Eng. Softw.*, 41, 886-892.

Tiryaki, B. (2008a). "Application of artificial neural networks for predicting the cuttability of rocks by drag tools." *Tunnelling Underground Space Technol.*, 23, 273-280.

Tiryaki, B. (2008b). "Predicting intact rock strength for mechanical excavation using multivariate statistics, artificial neural networks and regression trees." *Eng. Geol.*, 99, 51-60.

Tsuru, T. &Kozawa, T. (1998); "Noise characterization in SWD survey"; SocExplorGeophys Japan, Tokyo, Butsuri-Tansa; GeophysExplor;51(1):45-54.

Tsuru, T. and Kozawa, T. (1998). "Noise characterization in SWD survey." *Soc. Explor. Geophys. Japan*, Tokyo, Butsuri-Tansa, *Geophys. Explor.*, 51(1), 45–54.

Tugrul, A. and Zarif, I.H. (1999). "Correlation of mineralogical and textural characteristics with engineering properties of selected granitic rocks from Turkey." *Eng. Geol.*, 51, 303-317.

Ulusay, R. and Hudson, J.A., eds., (2007). "The complete ISRM suggested methods for rock characterization, testing and monitoring: 1974 – 2006." *Compilation arranged by the ISRM Turkish national group*, Ankara, Turkey, 628 p., ISBN: 978-975-93675-4-1

Vardhan, H. & Badal Yadav (2011); "An experimental study and analytical approach for estimating rock properties using sound levels from drilling"; Exploration and Production: Oil and Gas Review; Vol.9(2); pp.25-28.

Vardhan, H. & Murthy, Ch. S. N. (2007); "An experimental investigation of jack hammer drill noise with special emphasis on drilling in rocks of different compressive strengths"; Noise Control Engineering Journal – An International Publication; Vol. 55(3); pp. 282-293.

Vardhan, H. and Murthy, Ch.S.N. (2007). "An experimental investigation of jack hammer drill noise with special emphasis on drilling in rock s of different compressive strengths." *Noise Control Eng. J.*, 55(3), 282-293.

Vardhan, H., Adhikari, G. R. & Govinda Raj, M. (2009); "Estimating rock properties using sound levels produced during drilling"; International Journal of Rock Mechanics and Mining Sciences; Vol.46(3); pp. 604-612.

Vardhan, H., Adhikari, G.R. and Govindaraj, M. (2009). "Estimating rock properties using sound levels produced during drilling." *Int. J. Rock Mech. Min. Sci.*, 46(3), 604-612.

Verwaal, W. and Mulder, A. (1993). "Estimating rock strength with the Equotip hardness tester." *Int. J. Rock Mech. Min. Sci. Geomech. Abstr.*, 30, 659-662.

Vukelic, M.A. and Miranda, E.N. (1996). "Neural networks in petroleum engineering." *Int. J. Neural Syst.*, 7, 187.

Walker, A. (1963); "Noise-its effect and control in mining operations," The Canadian Mining and Metallurgical Bulletin, November, 820–834.

Ward, B. (1998). *German creek mines rock strength from velocity logs.* Unpublished report for Capricorn coal management pty ltd.

Williams, E. and Hogan, P.C. (2006). "Monitoring acoustic emission levels with changes in rock cutting conditions." *Pub., University of New South Wales,* www.mining.unsw.edu.au/Publications/publications_staff/Paper_WilliamsHogan_ AE_2006_web.htm, (July 2006).

Wong, T.F., David, C. and Zhu, W. (1997). "The transition from brittle faulting to cataclastic flow in porous sandstones: mechanical deformation." *J. Geophys. Res.*, 102, 3009-3025.

Yagiz, S., Sezer, E.A. and Gokceoglu, C. (2011). "Artificial neural networks and nonlinear regression techniques to assess the influence of slake durability cycles on the prediction of uniaxial compressive strength and modulus of elasticity for carbonate rocks." *Int. J. Numer. Anal. Methods Geomech.*, DOI: 10.1002/nag.1066.

Yang, Y. and Zhang, Q. (1998). "The applications of neural networks to rock engineering systems." *Int. J. Rock Mech. Min. Sci.*, 35(6), 727-745.

Yasar, E. and Erdogan, Y. (2004a). "Estimation of rock physicomechanical properties using hardness methods." *Eng. Geol.*, 71, 281-288.

Yasar, E. and Erdogan, Y. (2004b). "Correlating sound velocity with the density, compressive strength and Young's modulus of carbonate rocks." *Int. J. Rock Mech. Min. Sci.*, 41(5), 871-875.

Yavuz, H., Ugur, I. and Demirdag, S. (2008). "Abrasion resistance of carbonate rocks use in dimension stone industry and correlation between abrasion and rock properties." *Int. J. Rock Mech. Min. Sci.*, 45(2), 260-267.

Yilmaz, I. and Kaynar, O. (2011). "Multiple regression, ANN (RBF, MLP) and ANFIS models for prediction of swell potential of clayey soils." *Expert Syst. Appl.*, 38, 5958-5966.

Yilmaz, I. and Sendir, H. (2002). "Correlation of Schmidt hardness with unconfined compressive strength and Young's modulus in gypsum from Sivas (Turkey)." *Eng. Geol.*, 66, 211-219.

Yilmaz, I. and Yuksek, A.G. (2008). "An example of artificial neural network application for indirect estimation of rock parameters." *Rock Mech. Rock Eng.*, 41(5), 781-795.

Yilmaz, I. and Yuksek, A.G. (2009). "Prediction of the strength and elasticity modulus of gypsum using multiple regression, ANN, ANFIS models and their comparison." *Int. J. Rock Mech. Min. Sci.*, 46(4), 803-810.

Yuanyou, X., Yanming, X. and Ruigeng, Z. (1997). "An engineering geology evaluation method based on an artificial neural network and its application." *Eng. Geol.*, 47, 149-156.

Zborovjan, M. (2001). "Identification of minerals during drilling process via acoustic signal." *Metall. Foundry*, 26(4), Krakow, Poland.

Zborovjan, M., Lesso, I. and Dorcak, L. (2003). "Acoustic identification of rocks during drilling process." *J. Acta Montanistica Slovaca*, 8(4), 191-193.

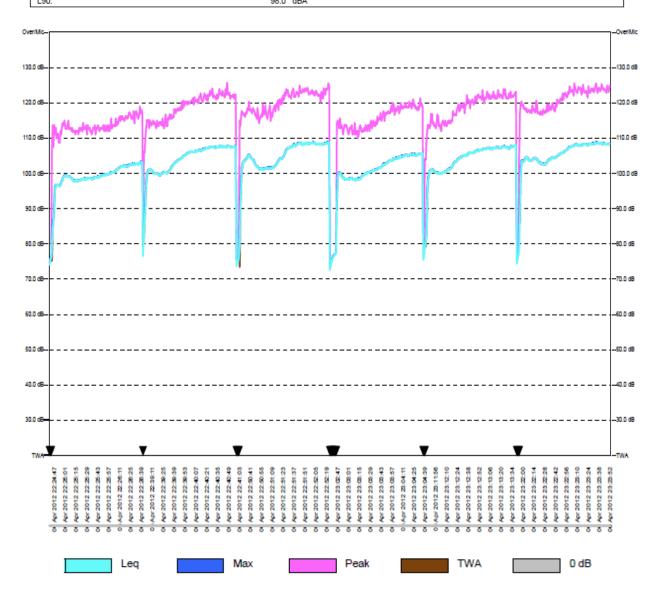
Zhang, L. (2005). *Engineering properties of rocks*. Elsevier Publications, Amsterdam, 4, 1-290.

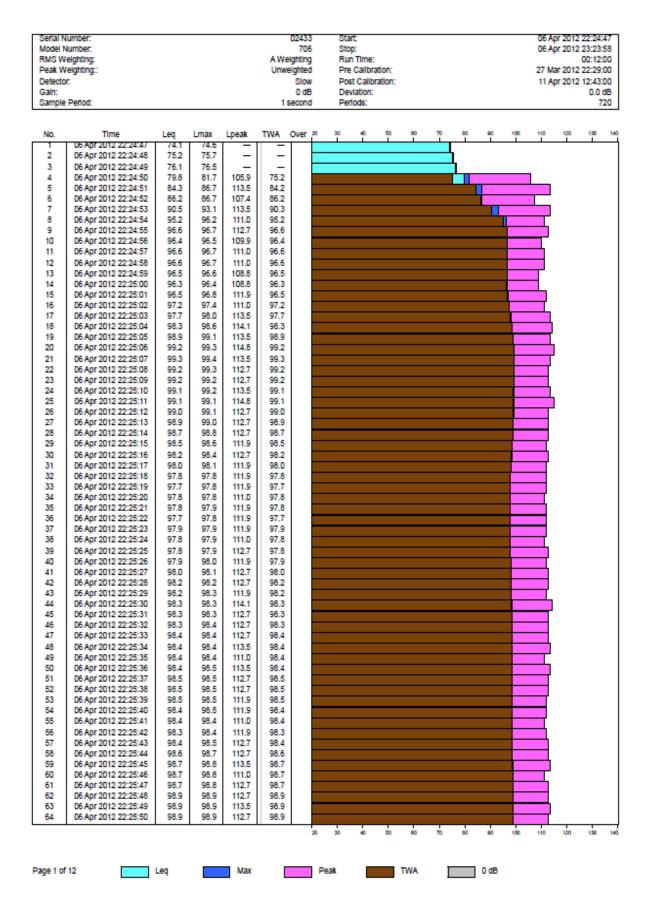
Zorlu, K., Gokceoglu, C., Ocakoglu, F., Nefeslioglu, H.A. and Acikalin, S. (2008). "Prediction of uniaxial compressive strength of sandstones using petrography based models." *Eng. Geol.*, 96, 141-158.

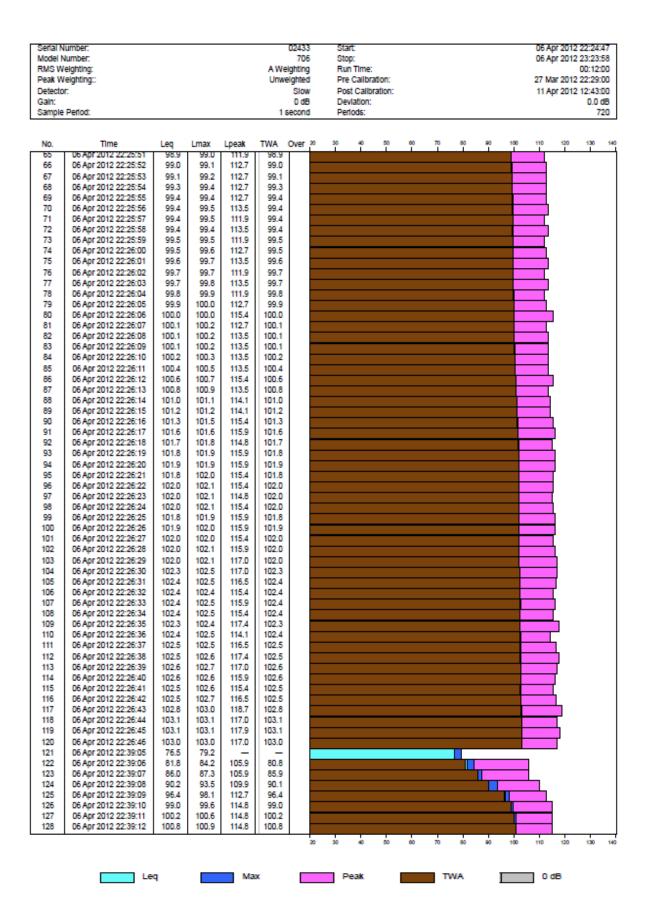
APPENDIX - I

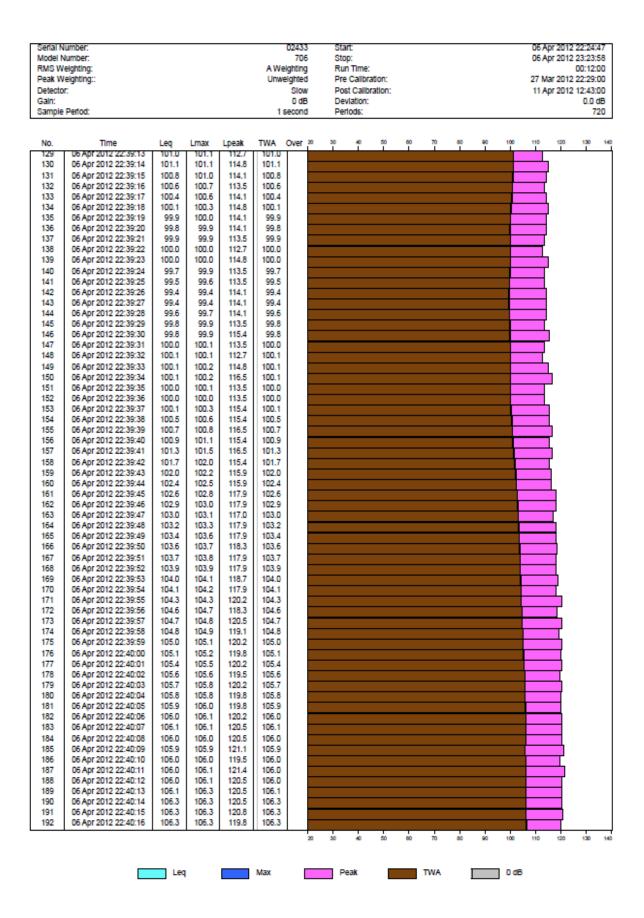
SOUND LEVELS FOR DRILLING IN THE IGNEOUS ROCK SAMPLES

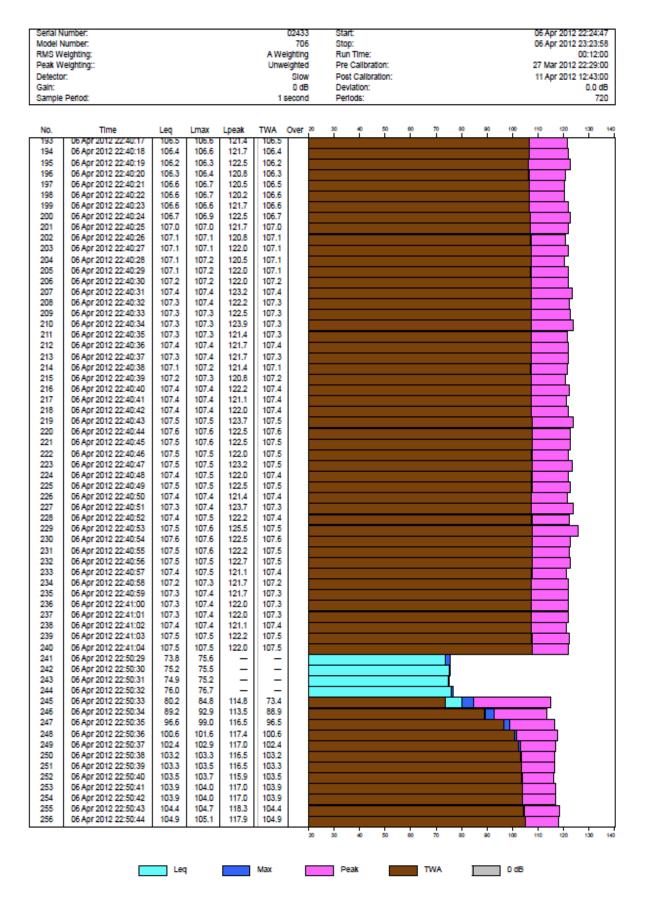
Serial Number:	02433	Start:	06 Apr 2012 22:24:47
Model Number:	706	Stop:	06 Apr 2012 23:23:58
RMS Weighting:	A Weighting	Run Time:	00:12:00
Peak Weighting::	Unweighted	Pre Calibration:	27 Mar 2012 22:29:00
Detector:	Slow	Post Calibration:	11 Apr 2012 12:43:00
Gain:	0 dB	Deviation:	0.0 dB
Sample Period:	1 second	Periods:	720
Exchange Rate:	5	Dose:	17.4 %
Threshold:	80.0 dBA	Projected Dose:	696.8 %
Criterion Level:	90.0 dBA	Leq:	104.6 dBA
Criterion Duration:	8.0 hours	TWA:	104.0 dBA
		TWA (8)	77.4 dBA
L10:	108.0 dBA	Lmax:	108.8 dBA
L30:	106.0 dBA	Lpeak (max):	125.5 dB
L50:	103.5 dBA	Lep (8)	88.6 dBA
L70:	100.5 dBA	SE:	2.3 Pa ² hr
L90:	98.0 dBA		

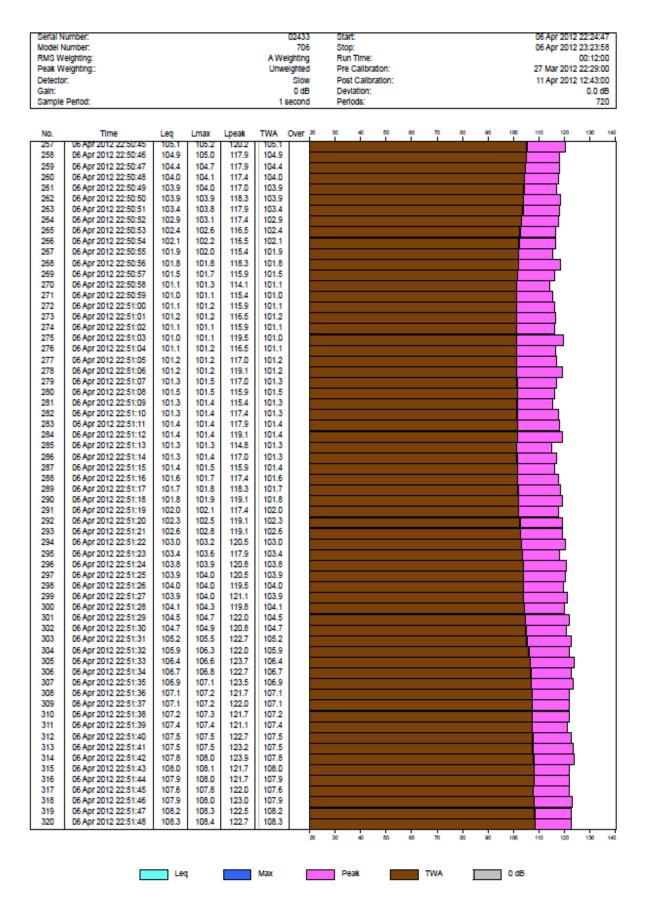




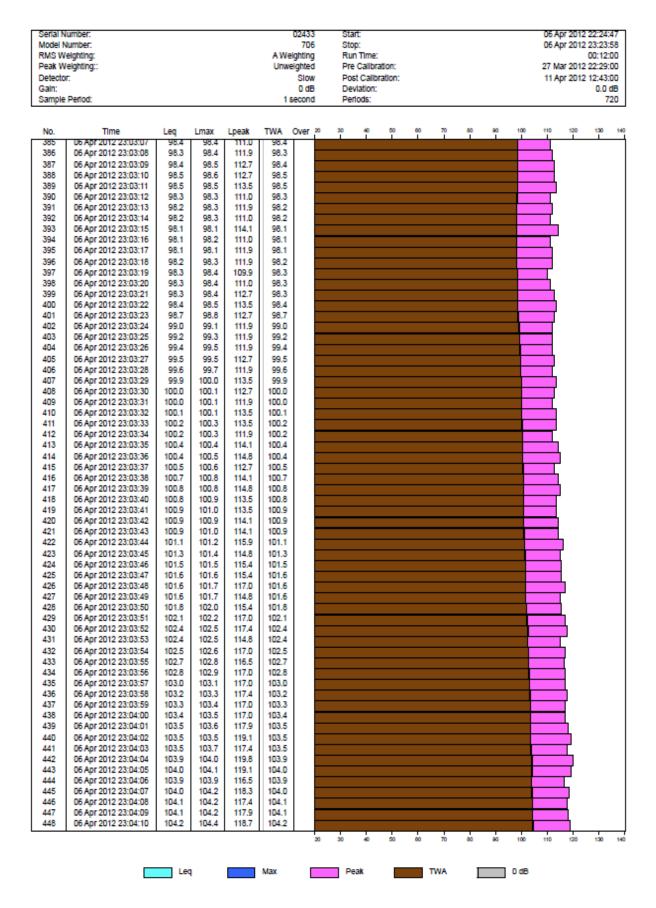


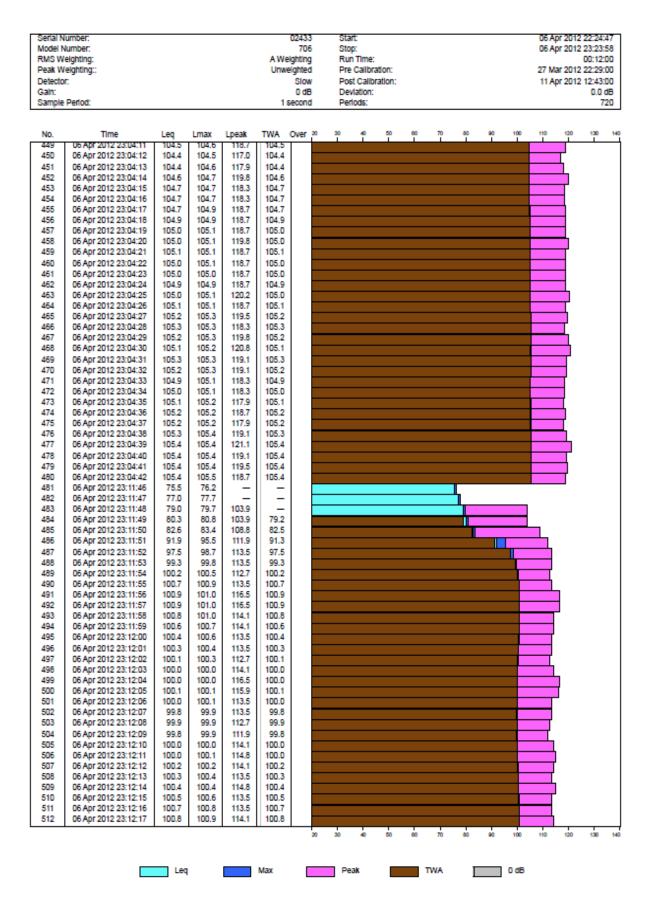




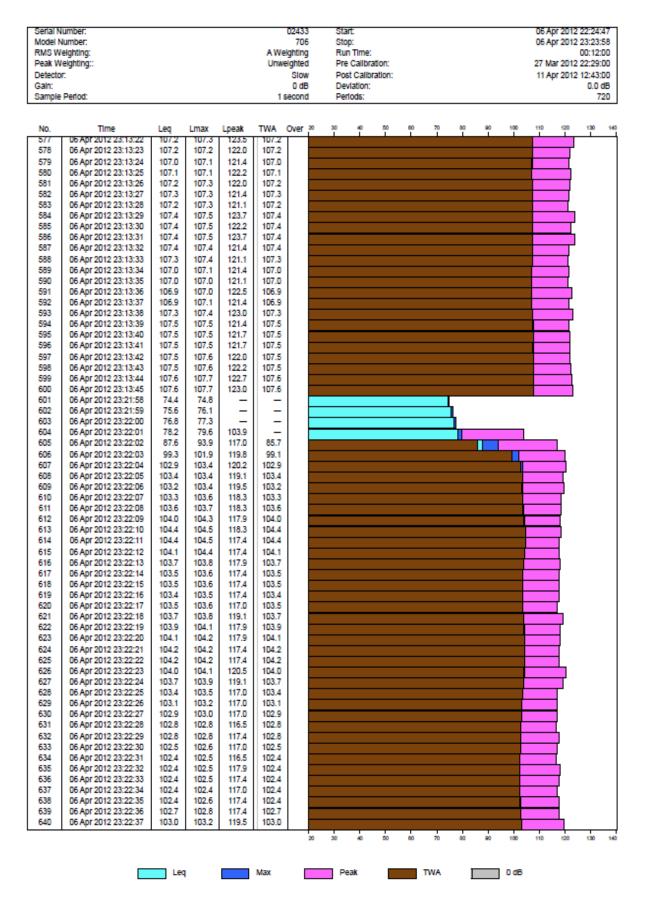


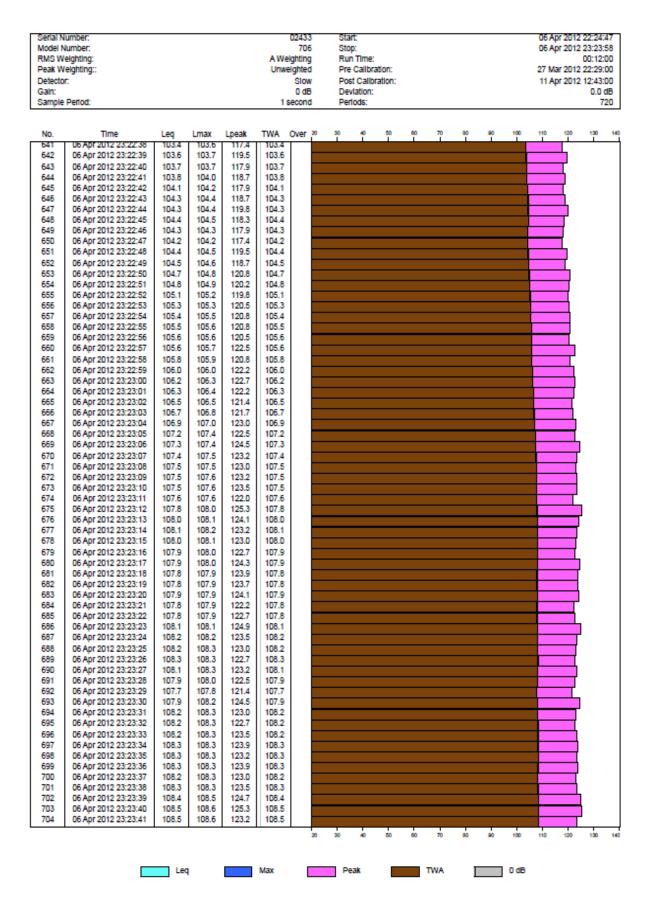
	lumber: leighting: leighting:: r.				Unwe	02433 706 Ighting Ighted Slow 0 dB second	5 Stop: 9 Run Time: 1 Pre Calibratio 9 Post Calibration: 8 Deviation:							06 Apr 20 27 Mar 20	12 22:24:47 12 23:23:58 00:12:00 12 22:29:00 12 12:43:00 0.0 dB 720
NO.	Time	Leq	Lmax	Lpeak	TWA	Over	20 30 40	50	60	70	80	90	100	110 120	130 140
321 322	06 Apr 2012 22:51:49 06 Apr 2012 22:51:50	108.3	108.4	121./ 124.7	108.3										
323	06 Apr 2012 22:51:50	108.5	108.6	124.7	108.5										
324	06 Apr 2012 22:51:52	108.3	108.4	123.0	108.3										\
325	06 Apr 2012 22:51:53	108.1	108.1	122.7	108.1										ť I
326	06 Apr 2012 22:51:54	108.1	108.1	123.7	108.1										
327	06 Apr 2012 22:51:55	108.2	108.3	123.2	108.2										
328 329	06 Apr 2012 22:51:56 06 Apr 2012 22:51:57	108.2	108.3	123.0 123.9	108.2										4
330	06 Apr 2012 22:51:58	108.4	108.5	123.2	108.4			_							-
331	06 Apr 2012 22:51:59	108.4	108.5	123.5	108.4										
332	06 Apr 2012 22:52:00	108.3	108.4	124.1	108.3										1
333	06 Apr 2012 22:52:01	108.3	108.4	123.2	108.3										J
334 335	06 Apr 2012 22:52:02	108.3	108.3	122.2	108.3										[]
335	06 Apr 2012 22:52:03 06 Apr 2012 22:52:04	108.3	108.4	122.7 122.5	108.3										4
337	06 Apr 2012 22:52:04	108.4	108.5	123.0	108.4										\
338	06 Apr 2012 22:52:06	108.5	108.5	123.5	108.5										1
339	06 Apr 2012 22:52:07	108.6	108.6	122.5	108.5										ſ I
340	06 Apr 2012 22:52:08	108.5	108.6	123.0	108.5]
341 342	06 Apr 2012 22:52:09 06 Apr 2012 22:52:10	108.4	108.5	122.0	108.4										
342	06 Apr 2012 22:52:10 06 Apr 2012 22:52:11	108.3	108.3	122.2	108.3										
344	06 Apr 2012 22:52:12	108.1	108.2	122.7	108.1										-
345	06 Apr 2012 22:52:13	108.1	108.2	122.2	108.1										' I
346	06 Apr 2012 22:52:14	108.2	108.2	122.0	108.2										·
347	06 Apr 2012 22:52:15	108.2	108.3	122.0	108.2										
348 349	06 Apr 2012 22:52:16 06 Apr 2012 22:52:17	108.2	108.3	122.0 120.2	108.2										
350	06 Apr 2012 22:52:18	108.1	108.2	123.0	108.1										ı
351	06 Apr 2012 22:52:19	108.2	108.2	121.7	108.2										
352	06 Apr 2012 22:52:20	108.3	108.3	121.1	108.3										
353	06 Apr 2012 22:52:21	108.2	108.3	123.9	108.2										
354 355	06 Apr 2012 22:52:22 06 Apr 2012 22:52:23	108.3 108.3	108.4	122.7	108.3)
356	06 Apr 2012 22:52:23	108.4	108.5	123.7	108.4										
357	06 Apr 2012 22:52:25	108.6	108.7	123.7	108.6			_							
358	06 Apr 2012 22:52:26	108.6	108.7	123.9	108.6										
359	06 Apr 2012 22:52:27	108.7	108.7	125.3	108.7										
360	06 Apr 2012 22:52:28	108.7	108.8	124.5	108.7										
361 362	06 Apr 2012 23:02:43 06 Apr 2012 23:02:44	72.6	73.2	_	_										
363	06 Apr 2012 23:02:44	74.6	74.9		=					_					
364	06 Apr 2012 23:02:46	75.4	75.8	-	-						1				
365	06 Apr 2012 23:02:47	76.2	76.5	-	-						1				
366	06 Apr 2012 23:02:48	76.5	76.6	-	-										
367 368	06 Apr 2012 23:02:49 06 Apr 2012 23:02:50	76.6	76.9	_							4				
360	06 Apr 2012 23:02:50 06 Apr 2012 23:02:51	77.3	77.6		1 -						4				
370	06 Apr 2012 23:02:52	91.4	96.6	114.1	89.5										
371	06 Apr 2012 23:02:53	98.7	99.7	114.8	98.7	1									
372	06 Apr 2012 23:02:54	100.0	100.1	113.5	100.0										
373	06 Apr 2012 23:02:55	100.1	100.1	112.7	100.1										
374 375	06 Apr 2012 23:02:56 06 Apr 2012 23:02:57	100.0	100.1 99.9	115.4 112.7	100.0										
376	06 Apr 2012 23:02:58	99.2	99.5	113.5	99.2	1									
377	06 Apr 2012 23:02:59	98.9	99.0	112.7	98.9	1									
378	06 Apr 2012 23:03:00	98.6	98.7	111.0	98.6										
379	06 Apr 2012 23:03:01	98.4	98.5	112.7	98.4										
380	06 Apr 2012 23:03:02	98.3	98.3	113.5	98.3	1									
381 382	06 Apr 2012 23:03:03 06 Apr 2012 23:03:04	98.3 98.3	98.4 98.3	113.5 113.5	98.3 98.3										
383	06 Apr 2012 23:03:04	98.3	98.3	112.7	98.3										
384	06 Apr 2012 23:03:06	98.2	98.3	111.9	98.2										
L	-					1	20 30 40	50	60	70	80	80	100	110 120	130 140
									_		_				
		Leo	9		Max	Γ	Peak			TWA	I		0 dB		
											-				





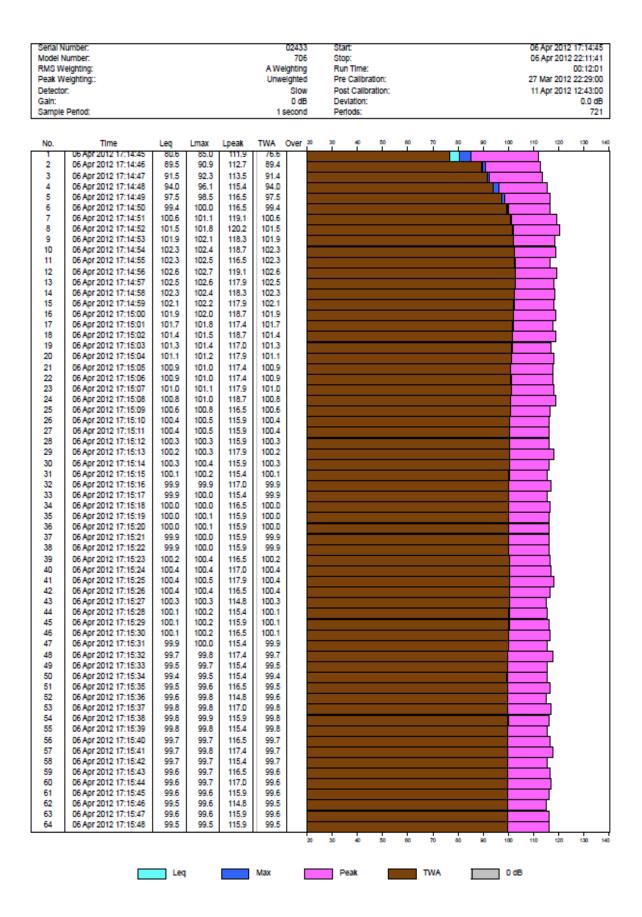
	łumber: leighting: feighting:: r:				Unwe	02433 706 Ighting Ighted Slow 0 dE second	5 Stop: g Run Time: d Pre Calibra v Post Calibr 3 Deviation:							06 A 27 M	pr 2012 ar 2012	22:24:47 23:23:58 00:12:00 22:29:00 12:43:00 0.0 dB 720
No.	Time	Leq	Lmax	Lpeak	TWA	Over	20 30 40	50	60	70	80	90	100	110	120	130 140
513 514	05 Apr 2012 23:12:18	101.0	101.2	114.8	101.0											
515	06 Apr 2012 23:12:19 06 Apr 2012 23:12:20	101.1	101.2	114.8	101.1								_		1	
516	06 Apr 2012 23:12:21	101.2	101.4	114.1	101.2											
517	06 Apr 2012 23:12:22	101.5	101.6	114.8	101.5											
518	06 Apr 2012 23:12:23	101.7	101.9	114.8	101.7										L	
519 520	06 Apr 2012 23:12:24	102.2	102.4	117.4 116.5	102.2										-	
520	06 Apr 2012 23:12:25 06 Apr 2012 23:12:26	102.7 103.0	102.9	116.5	102.7										-	
522	06 Apr 2012 23:12:27	103.2	103.3	116.5	103.2									_		
523	06 Apr 2012 23:12:28	103.3	103.3	116.5	103.3											
524	06 Apr 2012 23:12:29	103.4	103.5	117.9	103.4											
525	06 Apr 2012 23:12:30	103.7	103.9	118.3	103.7											
526 527	06 Apr 2012 23:12:31 06 Apr 2012 23:12:32	104.0 104.2	104.2	119.5 117.4	104.0											
528	06 Apr 2012 23:12:33	104.2	104.4	117.9	104.2										-	
529	06 Apr 2012 23:12:34	104.4	104.5	119.5	104.4											
530	06 Apr 2012 23:12:35	104.6	104.7	118.7	104.6											
531	06 Apr 2012 23:12:36	104.6	104.7	118.7	104.6											
532 533	06 Apr 2012 23:12:37 06 Apr 2012 23:12:38	104.8 104.9	104.9	119.5 119.8	104.8											
534	06 Apr 2012 23:12:39	104.9	103.0	117.0	104.9											
535	06 Apr 2012 23:12:40	105.0	105.1	119.1	105.0										-	
536	06 Apr 2012 23:12:41	105.0	105.0	119.5	105.0											
537	06 Apr 2012 23:12:42	105.1	105.1	122.0	105.1											
538 539	06 Apr 2012 23:12:43	105.2	105.3	120.2	105.2											
540	06 Apr 2012 23:12:44 06 Apr 2012 23:12:45	105.4	105.6	120.8 120.5	105.4										_	
541	06 Apr 2012 23:12:46	105.7	105.8	118.3	105.7										_	
542	06 Apr 2012 23:12:47	105.7	105.7	120.2	105.7											
543	06 Apr 2012 23:12:48	105.7	105.8	120.5	105.7											
544	06 Apr 2012 23:12:49	105.8	105.8	120.8	105.8											
545 546	06 Apr 2012 23:12:50 06 Apr 2012 23:12:51	105.7 105.8	105.8 105.8	119.8 119.5	105.7										_	
547	06 Apr 2012 23:12:52	105.9	105.0	120.5	105.9										_	
548	06 Apr 2012 23:12:53	106.1	106.2	121.7	106.1											
549	06 Apr 2012 23:12:54	106.2	106.3	120.2	106.2											
550	06 Apr 2012 23:12:55	106.2	106.2	121.1	106.2											
551 552	06 Apr 2012 23:12:56 06 Apr 2012 23:12:57	106.3 106.4	106.4	122.0 122.0	106.3											
553	06 Apr 2012 23:12:58	106.4	106.4	122.0	106.4											
554	06 Apr 2012 23:12:59	106.5	106.6	121.4	106.5											
555	06 Apr 2012 23:13:00	106.5	106.5	121.7	106.5											
556	06 Apr 2012 23:13:01	106.5	106.6	120.8	106.5											
557	06 Apr 2012 23:13:02	106.6	106.7	123.5	106.6											
558 559	06 Apr 2012 23:13:03 06 Apr 2012 23:13:04	106.7 106.8	106.8	121.1 119.8	106.7 106.8										_	
560	06 Apr 2012 23:13:05	106.7	106.8	121.1	106.7	1										
561	06 Apr 2012 23:13:06	106.8	106.8	121.1	106.8	1										
562	06 Apr 2012 23:13:07	106.9	106.9	121.4	106.9											
563 564	06 Apr 2012 23:13:08	107.0	107.1	123.0	107.0											
565	06 Apr 2012 23:13:09 06 Apr 2012 23:13:10	107.1	107.2	122.0	107.1											
566	06 Apr 2012 23:13:11	107.1	107.1	122.2	107.1											
567	06 Apr 2012 23:13:12	106.8	107.0	121.1	106.8											
568	06 Apr 2012 23:13:13	106.7	106.8	121.1	106.7	1										
569	06 Apr 2012 23:13:14	106.7	106.8	122.0	106.7	1										
570 571	06 Apr 2012 23:13:15 06 Apr 2012 23:13:16	106.9 107.0	107.0	121.7 123.0	106.9											
572	06 Apr 2012 23:13:17	107.1	107.1	123.0	107.1											
573	06 Apr 2012 23:13:18	107.1	107.2	122.0	107.1											
574	06 Apr 2012 23:13:19	107.1	107.2	121.1	107.1											
575	06 Apr 2012 23:13:20	107.3	107.3	121.1	107.3											
576	06 Apr 2012 23:13:21	107.2	107.2	121.4	107.2							-				
							20 30 40	50	60	70	80	90	100	110	120	130 140
	_	_					Der t				-					
		Lec	•		Max	l	Peak			TWA	1		0 dE			

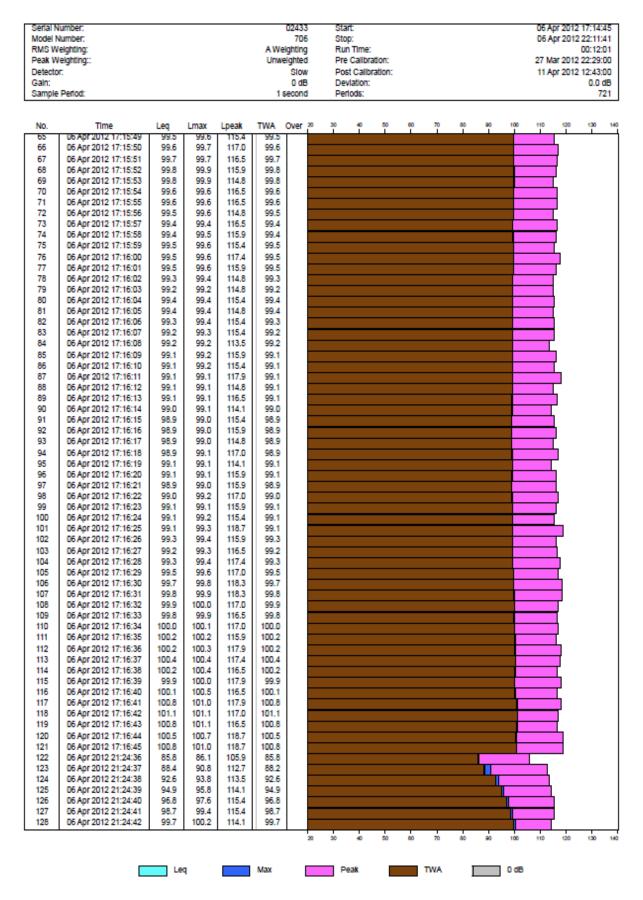


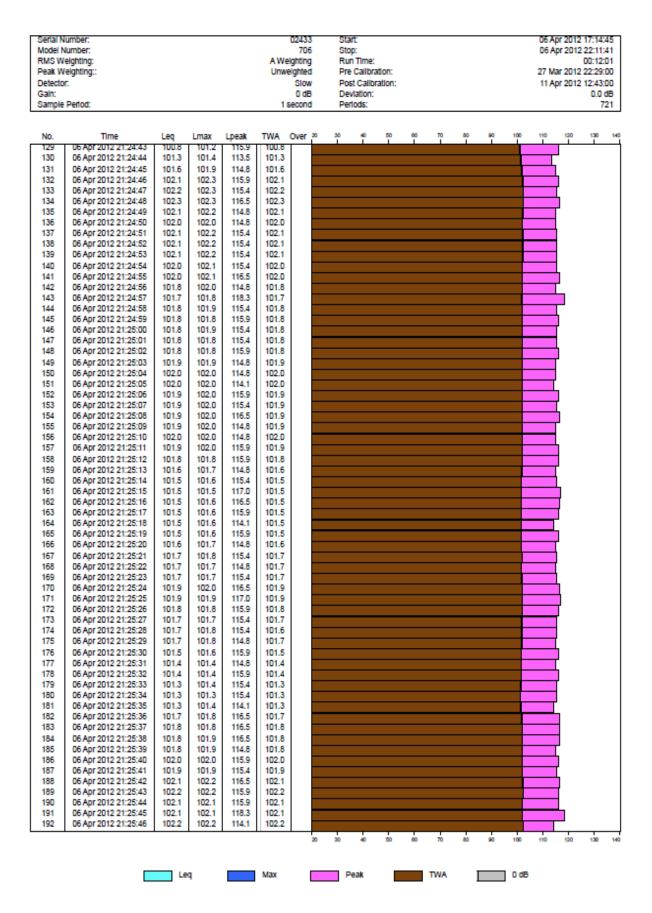


Serial Number:	02433	Start	06 Apr 2012 22:24:47
Model Number:	706	Stop:	06 Apr 2012 23:23:58
RMS Weighting:	A Weighting	Run Time:	00:12:00
Peak Weighting::	Unweighted	Pre Calibration:	27 Mar 2012 22:29:00
Detector:	Slow	Post Calibration:	11 Apr 2012 12:43:00
Gain:	0 dB	Deviation:	0.0 dB
Sample Period:	1 second	Periods:	720

NO.	Time	Leq	Lmax	Lpeak	TWA		20	30	40	50	60	70	80	90	100	110	120	130	140
705	06 Apr 2012 23:23:42	108.4	108.5	122.7	108.4														
706	06 Apr 2012 23:23:43	108.4	108.4	124.1	108.4														
707	06 Apr 2012 23:23:44	108.3	108.4	123.9	108.3														
708	06 Apr 2012 23:23:45	108.4	108.5	123.2	108.4														
709	06 Apr 2012 23:23:46	108.4	108.5	123.5	108.4														
710	06 Apr 2012 23:23:47	108.3	108.3	125.1	108.3													1	
711	06 Apr 2012 23:23:48	108.3	108.3	123.9	108.3														
712	06 Apr 2012 23:23:49	108.3	108.4	124.1	108.3														
713	06 Apr 2012 23:23:50	108.3	108.4	123.5	108.3														
714	06 Apr 2012 23:23:51	108.2	108.2	123.7	108.2														
715	06 Apr 2012 23:23:52	108.2	108.2	123.2	108.2														
716	06 Apr 2012 23:23:53	108.1	108.2	122.7	108.1														
717	06 Apr 2012 23:23:54	108.1	108.2	124.7	108.1														
718	06 Apr 2012 23:23:55	108.1	108.2	123.5	108.1														
719	06 Apr 2012 23:23:56	108.2	108.2	123.0	108.2														
720	06 Apr 2012 23:23:57	108.3	108.4	124.7	108.3														
							20	30		50	60	70	80	80	100	110	120	130	140
									-								-44		
		Leo	1		Max	[Pea	ik			TWA	Ι		0 dB				

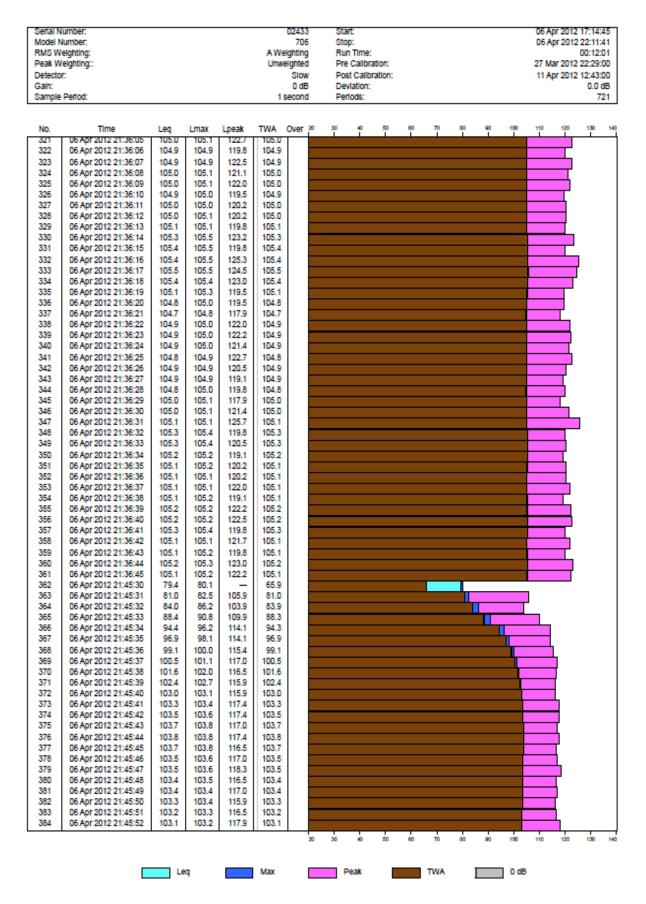




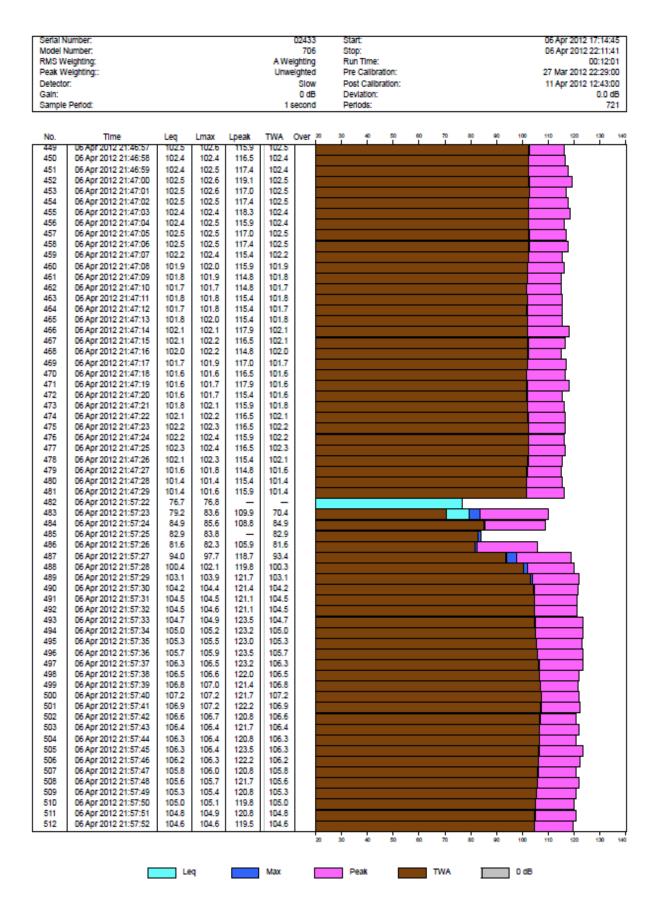


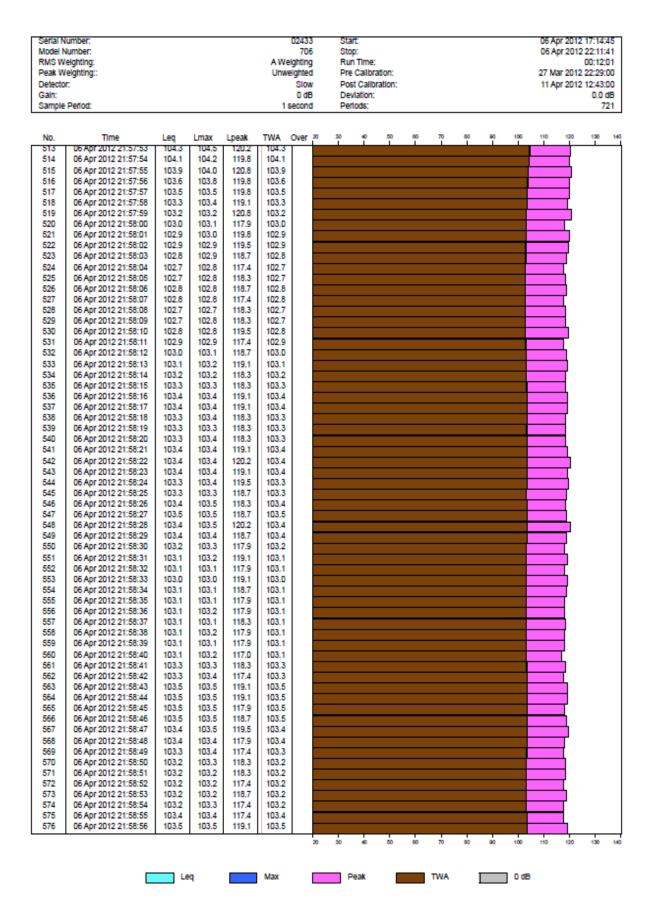
	Number: leighting: /eighting:: r:				Unwe	02433 706 Ighting Ighted Slow 0 dE second	5 Stop: 9 Run Time: 1 Pre Calibration: 9 Post Calibration 8 Deviation:					2	06 Apr 201 27 Mar 201	2 17:14:45 12 22:11:41 00:12:01 2 22:29:00 2 12:43:00 0.0 dB 721
No.	Time	Leq	Lmax	Lpeak		Over	20 30 40 50	60	70	80	90	100 1	10 120	130 140
193 194	06 Apr 2012 21:25:47 06 Apr 2012 21:25:48	102.2	102.3	115.9	102.2									
195	06 Apr 2012 21:25:49	101.9	101.9	115.9	101.9									
196	06 Apr 2012 21:25:50	102.0	102.1	115.9	102.0									
197	06 Apr 2012 21:25:51	102.1	102.1	115.9	102.1									
198 199	06 Apr 2012 21:25:52 06 Apr 2012 21:25:53	102.1	102.1	115.9 115.9	102.1							_		
200	06 Apr 2012 21:25:54	102.0	102.0	115.9	102.0							_		
201	06 Apr 2012 21:25:55	101.9	101.9	117.9	101.9									
202	06 Apr 2012 21:25:56	102.0	102.0	116.5	102.0								_	
203	06 Apr 2012 21:25:57	101.9	102.0	115.9	101.9									
204	06 Apr 2012 21:25:58	101.9	102.0	115.4	101.9									
205 206	06 Apr 2012 21:25:59 06 Apr 2012 21:26:00	101.9 101.8	101.9	115.4 116.5	101.9							_		
200	06 Apr 2012 21:26:00	101.0	102.0	115.9	101.0							_		
208	06 Apr 2012 21:26:02	102.0	102.1	115.9	102.0									
209	06 Apr 2012 21:26:03	102.2	102.4	116.5	102.2									
210	06 Apr 2012 21:26:04	102.3	102.4	117.9	102.3									
211	06 Apr 2012 21:26:05	102.2	102.3	115.4	102.2									
212 213	06 Apr 2012 21:26:06 06 Apr 2012 21:26:07	101.0	102.0	117.0 115.9	101.8							_		
213	06 Apr 2012 21:26:08	101.9	102.0	117.0	101.9								_	
215	06 Apr 2012 21:26:09	101.9	102.0	118.3	101.9									
216	06 Apr 2012 21:26:10	101.6	101.8	115.4	101.6									
217	06 Apr 2012 21:26:11	101.6	101.7	115.9	101.6									
218 219	06 Apr 2012 21:26:12 06 Apr 2012 21:26:13	101.7	101.7	115.9 117.0	101.7									
219	06 Apr 2012 21:26:13	101.6	101.0	114.8	101.7									
221	06 Apr 2012 21:26:15	101.6	101.7	117.0	101.6									
222	06 Apr 2012 21:26:16	101.5	101.6	115.4	101.5									
223	06 Apr 2012 21:26:17	101.2	101.4	115.4	101.2									
224	06 Apr 2012 21:26:18	101.3	101.3	115.9	101.3									
225 226	06 Apr 2012 21:26:19 06 Apr 2012 21:26:20	101.4	101.5	118.3 115.9	101.4							_		
227	06 Apr 2012 21:26:21	101.6	101.6	114.8	101.6								_	
228	06 Apr 2012 21:26:22	101.5	101.6	117.0	101.5									
229	06 Apr 2012 21:26:23	101.6	101.8	117.0	101.6									
230	06 Apr 2012 21:26:24	101.9	102.0	117.0	101.9									
231	06 Apr 2012 21:26:25	102.0	102.1	118.3	102.0									
232 233	06 Apr 2012 21:26:26 06 Apr 2012 21:26:27	102.1 102.3	102.2	116.5 118.3	102.1									
234	06 Apr 2012 21:26:28	102.3	102.3	117.0	102.3								_	
235	06 Apr 2012 21:26:29	102.2	102.3	115.9	102.2									
236	06 Apr 2012 21:26:30	102.1	102.2	115.9	102.1									
237	06 Apr 2012 21:26:31	102.0	102.2	117.9	102.0									
238 239	06 Apr 2012 21:26:32 06 Apr 2012 21:26:33	102.4	102.5	116.5 116.5	102.4									
239	06 Apr 2012 21:26:34	102.5	102.5	115.9	102.5	1								
241	06 Apr 2012 21:26:35	102.5	102.6	116.5	102.5									
242	06 Apr 2012 21:34:46	93.1	93.4	117.4	93.1									
243	06 Apr 2012 21:34:47	94.8	96.6	115.4	94.8									
244 245	06 Apr 2012 21:34:48 06 Apr 2012 21:34:49	98.7 101.6	100.2	117.4 118.3	98.7 101.6							_		
245	06 Apr 2012 21:34:50	101.0	102.5	118.7	101.0							_		
247	06 Apr 2012 21:34:51	103.6	103.9	119.1	103.6									
248	06 Apr 2012 21:34:52	104.0	104.1	119.5	104.0									
249	06 Apr 2012 21:34:53	104.1	104.2	119.5	104.1									
250	06 Apr 2012 21:34:54	104.0	104.1	117.4	104.0								Ţ	
251 252	06 Apr 2012 21:34:55 06 Apr 2012 21:34:55	104.1 104.4	104.1	118.3 119.1	104.1	1								
252	06 Apr 2012 21:34:56 06 Apr 2012 21:34:57	104.4	104.7	118.7	104.4									
254	06 Apr 2012 21:34:58	104.8	105.0	118.3	104.8									
255	06 Apr 2012 21:34:59	105.0	105.1	120.2	105.0									
256	06 Apr 2012 21:35:00	104.9	105.0	119.1	104.9									
							20 30 40 50	60	70	80	90	100 1	10 120	130 140
	_		eq		Max		Peak		TWA			0 dB		
			-1		-max		- CON							

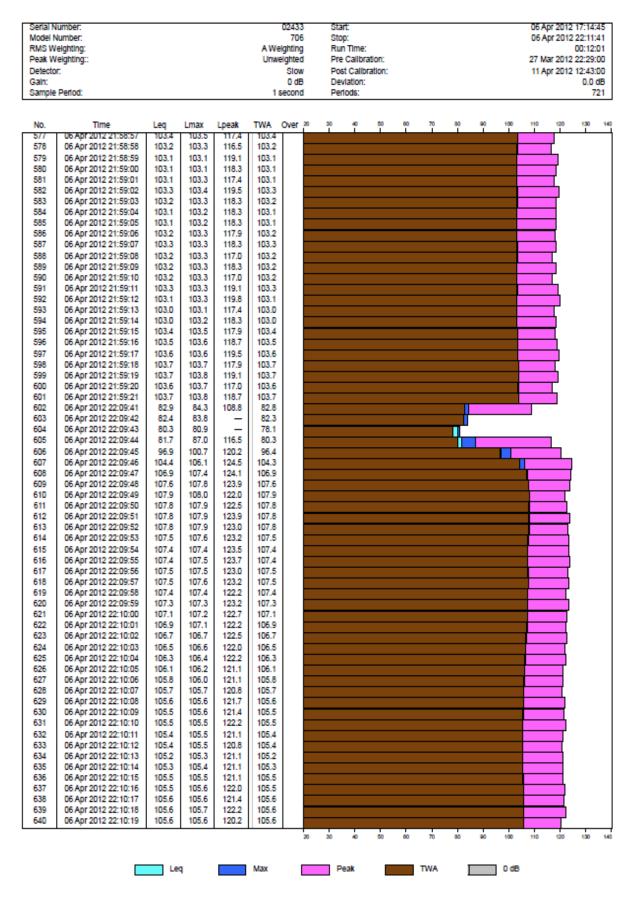
	lumber: eighting: leighting:: r:				Unwe	02433 706 Ighting Ighted Slow 0 dE second	Stop Run Pre Posi Dev): Time: Calibration t Calibratio lation:							06 Ap 27 Ma	or 2012 or 2012	17:14:45 22:11:41 00:12:01 22:29:00 12:43:00 0.0 dB 721	
No.	Time	Leq	Lmax	Lpeak	TWA	Over	20 30	40 S	0	60	70	80	90	100	110	120	130	140
257	06 Apr 2012 21:35:01	105.0	105.0	118.3	105.0											<u> </u>	_	1
258 259	06 Apr 2012 21:35:02	105.1	105.2	120.2 118.7	105.1													
260	06 Apr 2012 21:35:03 06 Apr 2012 21:35:04	105.2	105.5	120.2	105.2											4		
261	06 Apr 2012 21:35:05	105.8	106.1	121.1	105.8											-		
262	06 Apr 2012 21:35:06	106.3	106.4	120.5	106.3											-		
263	06 Apr 2012 21:35:07	106.5	106.6	121.7	106.5													
264	06 Apr 2012 21:35:08	106.6	106.7	120.5	106.6													
265 266	06 Apr 2012 21:35:09 06 Apr 2012 21:35:10	106.7	106.7	120.8 120.2	106.7										_	_		
267	06 Apr 2012 21:35:11	106.6	106.7	120.2	106.6													
268	06 Apr 2012 21:35:12	106.4	106.5	121.1	106.4											-		
269	06 Apr 2012 21:35:13	106.6	106.7	121.7	106.6													
270	06 Apr 2012 21:35:14	106.5	106.5	120.5	106.5													
271 272	06 Apr 2012 21:35:15 06 Apr 2012 21:35:16	106.2	106.3	119.5 120.8	106.2											_		
272	06 Apr 2012 21:35:16 06 Apr 2012 21:35:17	105.8	105.9	120.0	105.8											_		
274	06 Apr 2012 21:35:18	105.8	105.9	120.2	105.8										-	-		
275	06 Apr 2012 21:35:19	105.7	105.8	119.5	105.7													
276	06 Apr 2012 21:35:20	105.7	105.7	120.2	105.7													
277	06 Apr 2012 21:35:21	105.7	105.7	119.1	105.7											_		
278 279	06 Apr 2012 21:35:22 06 Apr 2012 21:35:23	105.6	105.6	119.8 119.1	105.6											_		
2/9	06 Apr 2012 21:35:24	105.5	105.5	119.8	105.5											4		
281	06 Apr 2012 21:35:25	105.2	105.4	118.3	105.2											-		
282	06 Apr 2012 21:35:26	105.0	105.0	120.2	105.0													
283	06 Apr 2012 21:35:27	104.9	105.0	118.3	104.9													
284 285	06 Apr 2012 21:35:28	104.6 104.5	104.7	118.7 117.4	104.6											-		
205	06 Apr 2012 21:35:29 06 Apr 2012 21:35:30	104.5	104.5	119.1	104.5											4		
287	06 Apr 2012 21:35:31	104.5	104.5	118.7	104.0											-		
288	06 Apr 2012 21:35:32	104.5	104.5	117.4	104.5													
289	06 Apr 2012 21:35:33	104.5	104.5	118.7	104.5													
290 291	06 Apr 2012 21:35:34	104.6	104.6	118.3	104.6											Ц.		
291	06 Apr 2012 21:35:35 06 Apr 2012 21:35:36	104.5 104.5	104.5	119.5 117.9	104.5											-		
293	06 Apr 2012 21:35:37	104.5	104.6	119.8	104.5									_		-		
294	06 Apr 2012 21:35:38	104.5	104.6	118.7	104.5											-		
295	06 Apr 2012 21:35:39	104.7	104.7	118.7	104.7													
296	06 Apr 2012 21:35:40	104.6	104.7	117.9	104.6													
297 298	06 Apr 2012 21:35:41 06 Apr 2012 21:35:42	104.6 104.6	104.7	117.9 118.7	104.6											4		
290	06 Apr 2012 21:35:42 06 Apr 2012 21:35:43	104.6	104.7	117.9	104.6											-		
300	06 Apr 2012 21:35:44	104.6	104.6	118.7	104.6											4		
301	06 Apr 2012 21:35:45	104.4	104.5	117.9	104.4											1		
302	06 Apr 2012 21:35:46	104.3	104.4	118.7	104.3													
303	06 Apr 2012 21:35:47	104.4	104.4	117.4	104.4											L,		
304 305	06 Apr 2012 21:35:48 06 Apr 2012 21:35:49	104.4 104.5	104.5	118.7 118.7	104.4	1												
305	06 Apr 2012 21:35:50	104.5	104.6	117.4	104.6													
307	06 Apr 2012 21:35:51	104.7	104.7	118.3	104.7											1		
308	06 Apr 2012 21:35:52	104.6	104.7	119.1	104.6	1												
309	06 Apr 2012 21:35:53	104.6	104.7	122.0	104.6													
310 311	06 Apr 2012 21:35:54 06 Apr 2012 21:35:55	104.7	104.7	120.8 118.3	104.7													
312	06 Apr 2012 21:35:55	104.7	104.7	119.8	104.7	1										-		
313	06 Apr 2012 21:35:57	105.0	105.0	121.4	105.0													
314	06 Apr 2012 21:35:58	104.9	105.1	120.2	104.9													
315	06 Apr 2012 21:35:59	105.0	105.1	120.5	105.0	1												
316	06 Apr 2012 21:36:00	105.1	105.1	120.2	105.1													
317 318	06 Apr 2012 21:36:01 06 Apr 2012 21:36:02	105.1	105.1	118.7 122.2	105.1													
319	06 Apr 2012 21:36:02	103.0	103.1	122.2	104.9	1												
320	06 Apr 2012 21:36:04	104.9	105.0	123.2	104.9													
	-	-	-				20 30	a 5	0	60	70	80	90	100	110	120	130	140
		Le	P		Max		P	eak		Т	WA	Ι		0 dB				

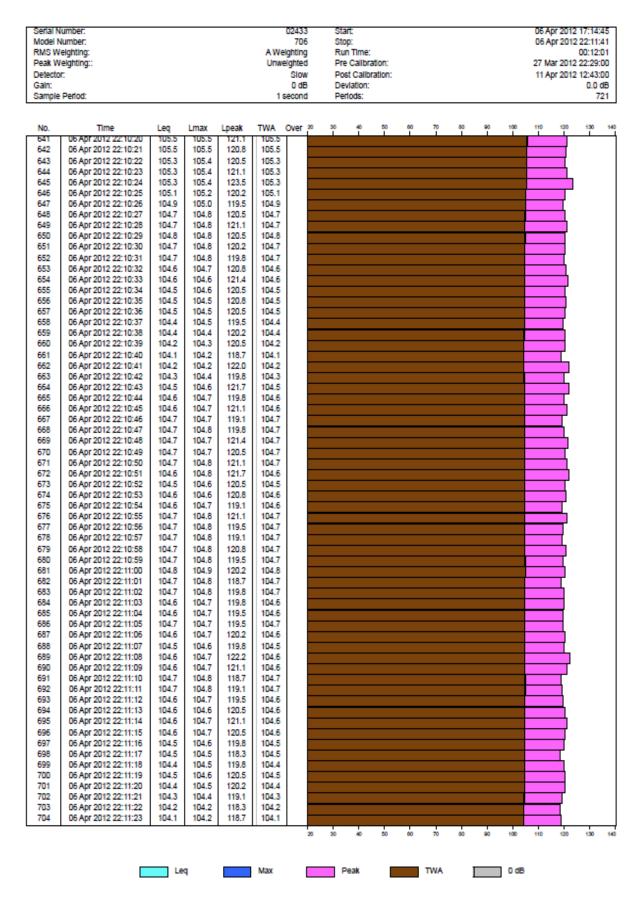


	Number: leighting: feighting:: r:				Unwe	02433 706 Ighting Ighted Slow 0 dE second	Sto Ru Pro Po De								06 Ap 27 Ma	r 2012 or 2012 r 2012 r 2012	22:11) 00:12) 22:29) 12:43) 0.0 (:41 :01 :00
No.	Time	Leq	Lmax	Lpeak	TWA	Over	20 30	-	50	60	70	80	90	100	110	120	130	140
385	06 Apr 2012 21:45:53	102.9	103.0	117.0	102.9													-
386	06 Apr 2012 21:45:54	102.6	102.8	115.9	102.6											ſ		
387	06 Apr 2012 21:45:55	102.3	102.4	115.9	102.3											Į –		
388	06 Apr 2012 21:45:56 06 Apr 2012 21:45:57	102.1 102.0	102.2	116.5 115.4	102.1													
390	06 Apr 2012 21:45:58	102.0	102.1	117.0	102.0									_		•		
391	06 Apr 2012 21:45:59	102.0	102.1	116.5	102.1									_		ł		
392	06 Apr 2012 21:46:00	102.1	102.2	114.8	102.1													
393	06 Apr 2012 21:46:01	102.2	102.3	117.4	102.2													
394	06 Apr 2012 21:46:02	102.2	102.3	115.9	102.2											Γ		
395	06 Apr 2012 21:46:03	102.1	102.2	115.9	102.1											L		
396	06 Apr 2012 21:46:04	102.0	102.1	117.4	102.0									_				
397 398	06 Apr 2012 21:46:05 06 Apr 2012 21:46:06	101.8 101.8	101.9	115.9 115.9	101.8									_				
399	06 Apr 2012 21:46:07	101.8	101.9	115.9	101.8									_				
400	06 Apr 2012 21:46:08	101.6	101.7	115.4	101.6													
401	06 Apr 2012 21:46:09	101.5	101.5	115.9	101.5													
402	06 Apr 2012 21:46:10	101.5	101.6	117.4	101.5													
403	06 Apr 2012 21:46:11	101.6	101.8	116.5	101.6											ſ		
404	06 Apr 2012 21:46:12	101.9	101.9	115.9	101.9											Ĺ		
405 406	06 Apr 2012 21:46:13 06 Apr 2012 21:46:14	102.0	102.0	117.0 116.5	102.0									_		ļ		
400	06 Apr 2012 21:46:14	102.2	102.2	116.5	102.2									_				
408	06 Apr 2012 21:46:16	102.3	102.3	117.4	102.3											4		
409	06 Apr 2012 21:46:17	102.3	102.3	119.5	102.3											-		
410	06 Apr 2012 21:46:18	102.3	102.4	118.3	102.3													
411	06 Apr 2012 21:46:19	102.2	102.3	118.3	102.2													
412	06 Apr 2012 21:46:20	102.1	102.2	117.4	102.1													
413	06 Apr 2012 21:46:21 06 Apr 2012 21:46:22	101.9	102.0	120.2 115.9	101.9									_				
414	06 Apr 2012 21:46:23	101.9	102.0	119.1	101.9									_				
416	06 Apr 2012 21:46:24	102.4	102.4	118.7	102.4											-		
417	06 Apr 2012 21:46:25	102.4	102.4	117.9	102.4											7		
418	06 Apr 2012 21:46:26	102.3	102.3	117.0	102.3											Ľ		
419	06 Apr 2012 21:46:27	102.3	102.4	118.3	102.3													
420 421	06 Apr 2012 21:46:28	102.5	102.5	117.4 120.2	102.5													
421	06 Apr 2012 21:46:29 06 Apr 2012 21:46:30	102.5	102.6	119.5	102.5											_		
423	06 Apr 2012 21:46:31	102.4	102.4	118.3	102.4									_		-		
424	06 Apr 2012 21:46:32	102.3	102.4	117.0	102.3													
425	06 Apr 2012 21:46:33	102.2	102.3	116.5	102.2											ſ		
426	06 Apr 2012 21:46:34	102.4	102.5	115.9	102.4											Ĺ		
427	06 Apr 2012 21:46:35	102.5	102.5	119.1	102.5													
428 429	06 Apr 2012 21:46:36 06 Apr 2012 21:46:37	102.5	102.6	116.5 116.5	102.5													
430	06 Apr 2012 21:46:38	102.8	102.8	117.0	102.8											4		
431	06 Apr 2012 21:46:39	102.7	102.8	117.0	102.7													
432	06 Apr 2012 21:46:40	102.6	102.7	115.9	102.6											ſ		
433	06 Apr 2012 21:46:41	102.6	102.7	117.4	102.6											1		
434	06 Apr 2012 21:46:42	102.7	102.8	118.3	102.7													
435 436	06 Apr 2012 21:46:43 06 Apr 2012 21:46:44	102.8	102.8	118.3 118.3	102.8											_		
437	06 Apr 2012 21:46:45	102.9	103.0	118.3	102.9											-		
438	06 Apr 2012 21:46:46	102.8	103.0	116.5	102.8													
439	06 Apr 2012 21:46:47	102.8	102.8	118.7	102.8											5		
440	06 Apr 2012 21:46:48	102.9	102.9	117.9	102.9													
441	06 Apr 2012 21:46:49	102.9	102.9	117.0	102.9											Ţ		
442 443	06 Apr 2012 21:46:50 06 Apr 2012 21:46:51	102.9	102.9	117.4 117.9	102.9	1										4		
44-5	06 Apr 2012 21:46:52	102.0	102.9	117.0	102.0											-		
445	06 Apr 2012 21:46:53	102.0	102.8	115.9	102.7											ł		
446	06 Apr 2012 21:46:54	102.6	102.6	117.9	102.6											5		
447	06 Apr 2012 21:46:55	102.6	102.7	116.5	102.6											Ľ		
448	06 Apr 2012 21:46:56	102.6	102.6	117.0	102.6													
							20 30	4	50	60	70	80	90	100	110	120	130	140
		Le	eq		Мах			Peak			TWA			0 d	в			



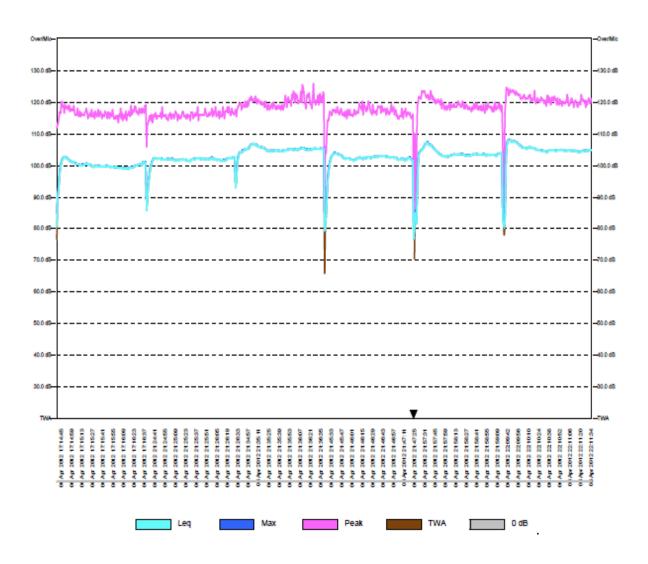




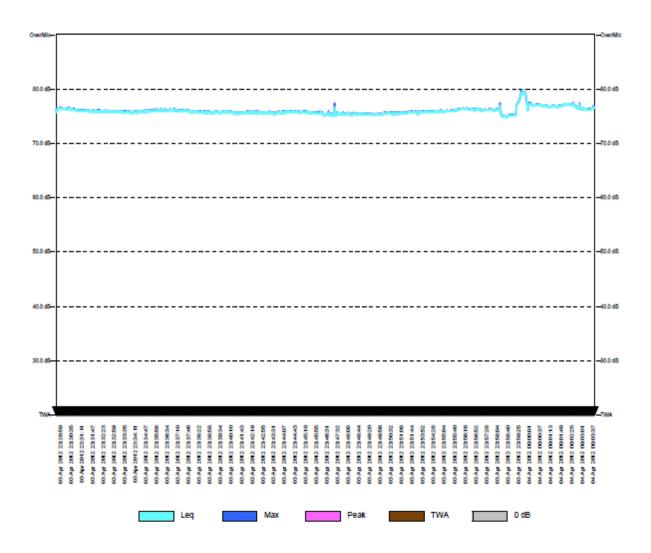


	lumber: leighting: leighting:: r:				Unw	0243 70 eightin veightei Slov 0 di secon	6 g d N B	Start: Stop: Run Time Pre Calibi Post Caliti Deviation: Periods:	ration: pration:						06 A	pr 2012 pr 2012 ar 2012 pr 2012	22:11 00:12 22:29 12:43 0.0	:41 :01 :00
No.	Time	Leg	Lmax	Lpeak	TWA	Over	20 30		50	60	70	80	90	100	110	120	130	140
705	05 ADF 2012 22:11:24	104.1	104.2	119.8	104.1											-		
706	06 Apr 2012 22:11:25	104.2	104.2	119.8	104.2											_		
707	05 Apr 2012 22:11:25	104.3	104.3	119.1	104.3											-		
708	06 Apr 2012 22:11:27	104.4	104.4	122.2	104.4													
709	06 Apr 2012 22:11:28	104.5	104.5	120.5	104.5											_		
710	06 Apr 2012 22:11:29	104.4	104.4	120.8	104.4													
711	06 Apr 2012 22:11:30	104.5	104.5	120.2	104.5													
712	06 Apr 2012 22:11:31	104.6	104.6	120.2	104.6													
713	06 Apr 2012 22:11:32	104.6	104.7	120.5	104.6													
714	06 Apr 2012 22:11:33	104.5	104.5	118.3	104.5													
715	06 Apr 2012 22:11:34	104.4	104.5	120.2	104.4													
716	06 Apr 2012 22:11:35	104.5	104.5	119.1	104.5													
717	06 Apr 2012 22:11:36	104.5	104.6	120.5	104.5													
718	06 Apr 2012 22:11:37	104.6	104.8	121.1	104.6													
719	06 Apr 2012 22:11:38	104.8	104.8	119.8	104.8													
720	06 Apr 2012 22:11:39	104.7	104.8	119.5	104.7													
721	06 Apr 2012 22:11:40	104.7	104.8	120.8	104.7													
							20 30	•	50	60	70	80	90	100	110	120	130	140
		Le	q		Мах			Peak			TWA]] 0 dB				

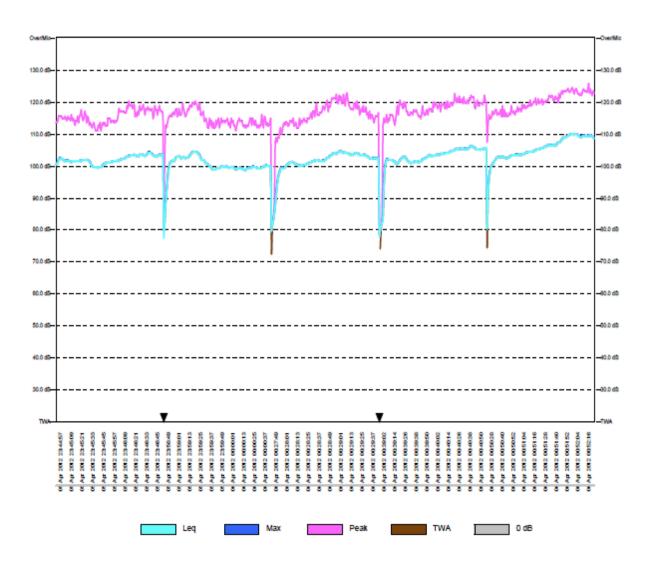
Serial Number:	02433	Start:	05 Apr 2012 17:14:45
Model Number:	706	Stop:	06 Apr 2012 22:11:41
RMS Weighting:	A Weighting	Run Time:	00:12:01
Peak Weighting::	Unweighted	Pre Calibration:	27 Mar 2012 22:29:00
Detector:	Slow	Post Calibration:	11 Apr 2012 12:43:00
Gain:	0 dB	Deviation:	0.0 dB
Sample Period:	1 second	Periods:	721
Exchange Rate:	5	Dose:	15.2 %
Threshold:	80.0 dBA	Projected Dose:	608.3 %
Criterion Level:	90.0 dBA	Leq:	103.3 dBA
Criterion Duration:	8.0 hours	TWA:	103.0 dBA
		TWA (8)	76.4 dBA
L10:	105.0 dBA	Lmax	108.0 dBA
L30:	104.0 dBA	Lpeak (max):	125.7 dB
L50:	102.5 dBA	Lep (8)	87.3 dBA
L70:	101.5 dBA	SE:	1.7 Pa ^a hr
L90:	99.5 dBA		



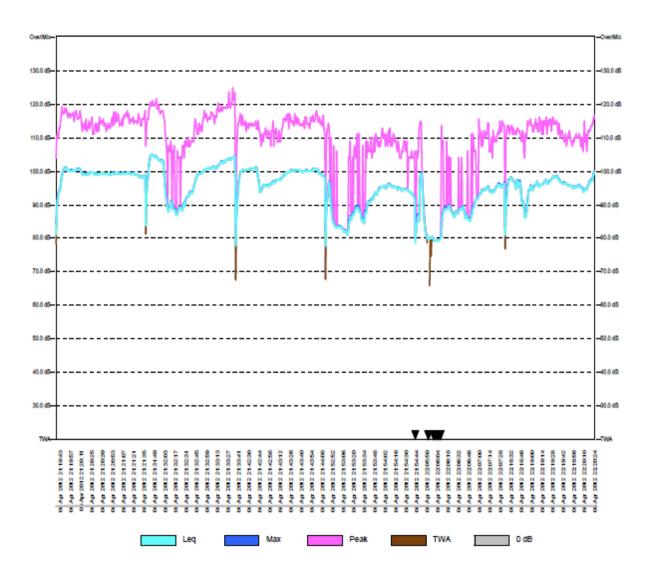
Serial Number:	02433	Start	03 Apr 2012 23:29:59
Model Number:	706	Stop:	04 Apr 2012 00:03:58
RMS Weighting:	A Weighting	Run Time:	00:30:21
Peak Weighting::	Unweighted	Pre Calibration:	27 Mar 2012 22:29:00
Detector:	Slow	Post Calibration:	11 Apr 2012 12:43:00
Gain:	0 dB	Deviation:	0.0 dB
Sample Period:	1 second	Periods:	1821
Exchange Rate:	5	Dose:	-
Threshold:	80.0 dBA	Projected Dose:	_
Criterion Level:	90.0 dBA	Leq:	76.0 dBA
Criterion Duration:	8.0 hours	TWA:	-
		TWA (8)	_
L10:	76.5 dBA	Lmax	79.8 dBA
L30:	76.0 dBA	Lpeak (max):	-
L50:	75.5 dBA	Lep (8)	64.0 dBA
L70:	75.5 dBA	SE:	0.0 Pa ^a hr
L90:	75.0 dBA		



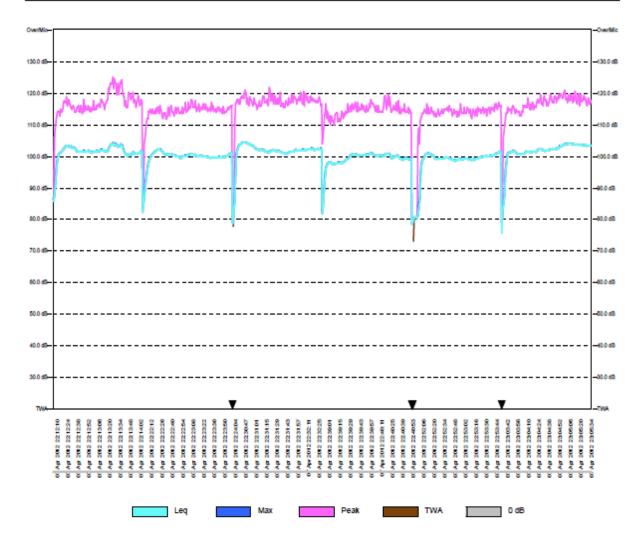
Serial Number:	02433	Start:	05 Apr 2012 23:44:57
Model Number:	706	Stop:	06 Apr 2012 00:52:28
RMS Weighting:	A Weighting	Run Time:	00:10:00
Peak Weighting::	Unweighted	Pre Calibration:	27 Mar 2012 22:29:00
Detector:	Slow	Post Calibration:	11 Apr 2012 12:43:00
Gain:	0 dB	Devlation:	0.0 dB
Sample Period:	1 second	Periods:	600
Exchange Rate:	5	Dose:	12.6 %
Threshold:	80.0 dBA	Projected Dose:	603.9 %
Criterion Level:	90.0 dBA	Leq:	103.4 dBA
Criterion Duration:	8.0 hours	TWA:	103.0 dBA
L10:	105.5 dBA	TWA (8)	75.0 dBA
L30:	103.5 dBA	Lmax	110.0 dBA
L50:	102.0 dBA	Lpeak (max):	125.7 dB
L70:	101.0 dBA	Lep (8)	86.6 dBA
L90:	99.0 dBA	SE:	1.5 PaPhr



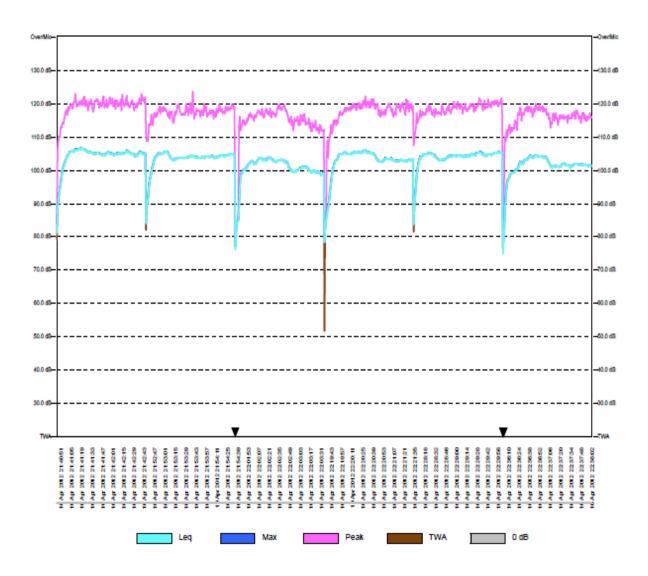
Serial Number:	02433	Start	08 Apr 2012 21:19:43
Model Number:	706	Stop:	08 Apr 2012 22:20:30
RMS Weighting:	A Weighting	Run Time:	00:12:00
Peak Weighting::	Unweighted	Pre Calibration:	27 Mar 2012 22:29:00
Detector:	Slow	Post Calibration:	11 Apr 2012 12:43:00
Gain:	0 dB	Devlation:	0.0 dB
Sample Period:	1 second	Periods:	720
Exchange Rate:	5	Dose:	6.3 %
Threshold:	80.0 dBA	Projected Dose:	252.7 %
Criterion Level:	90.0 dBA	Leq:	97.5 dBA
Criterion Duration:	8.0 hours	TWA:	96.7 dBA
		TWA (8)	70.1 dBA
L10:	100.5 dBA	Lmax	104.9 dBA
L30:	99.0 dBA	Lpeak (max):	124.7 dB
L50:	95.5 dBA	Lep (8)	81.5 dBA
L70:	93.5 dBA	SE:	0.5 Pa ^s hr
L90:	85.5 dBA		



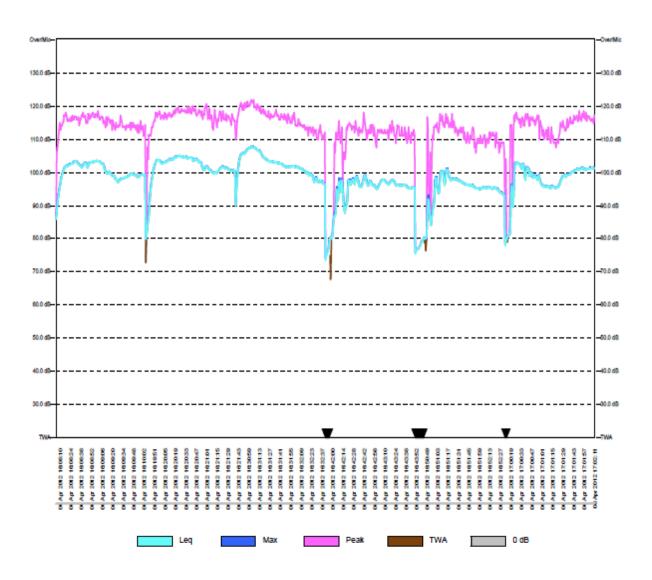
Serial Number:	02433	Start:	07 Apr 2012 22:12:10
Model Number:	706	Stop:	07 Apr 2012 23:05:40
RMS Weighting:	A Weighting	Run Time:	00:12:00
Peak Weighting::	Unweighted	Pre Calibration:	27 Mar 2012 22:29:00
Detector:	Slow	Post Calibration:	11 Apr 2012 12:43:00
Gain:	0 dB	Deviation:	0.0 dB
Sample Period:	1 second	Periods:	720
Exchange Rate:	5	Dose:	11.0 %
Threshold:	80.0 dBA	Projected Dose:	439.7 %
Criterion Level:	90.0 dBA	Leq:	100.9 dBA
Criterion Duration:	8.0 hours	TWA:	100.7 dBA
		TWA (8)	74.1 dBA
L10:	103.0 dBA	Lmax	104.4 dBA
L30:	101.5 dBA	Lpeak (max):	124.9 dB
L50:	100.5 dBA	Lep (8)	84.9 dBA
L70:	99.5 dBA	SE:	1.0 Paħr
L90:	98.0 dBA		



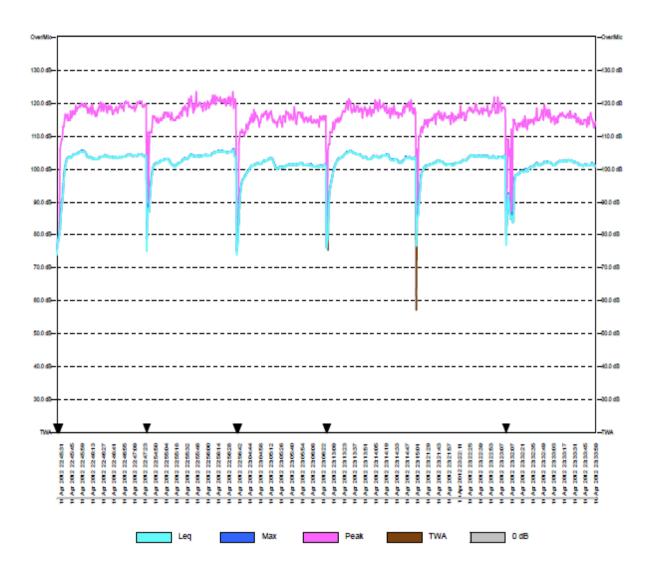
Serial Number:	02433	Start	10 Apr 2012 21:40:51
Model Number:	706	Stop:	10 Apr 2012 22:38:08
RMS Weighting:	A Weighting	Run Time:	00:12:00
Peak Weighting::	Unweighted	Pre Calibration:	27 Mar 2012 22:29:00
Detector:	Slow	Post Calibration:	11 Apr 2012 12:43:00
Gain:	0 dB	Deviation:	0.0 dB
Sample Period:	1 second	Periods:	720
Exchange Rate:	5	Dose:	15.5 %
Threshold:	80.0 dBA	Projected Dose:	619.7 %
Criterion Level:	90.0 dBA	Leq:	103.4 dBA
Criterion Duration:	8.0 hours	TWA:	103.2 dBA
		TWA (8)	76.5 dBA
L10:	105.0 dBA	Lmax	106.4 dBA
L30:	104.5 dBA	Lpeak (max):	123.5 dB
L50:	103.5 dBA	Lep (8)	87.4 dBA
L70:	102.0 dBA	SE:	1.7 Paªhr
L90:	99.0 dBA		

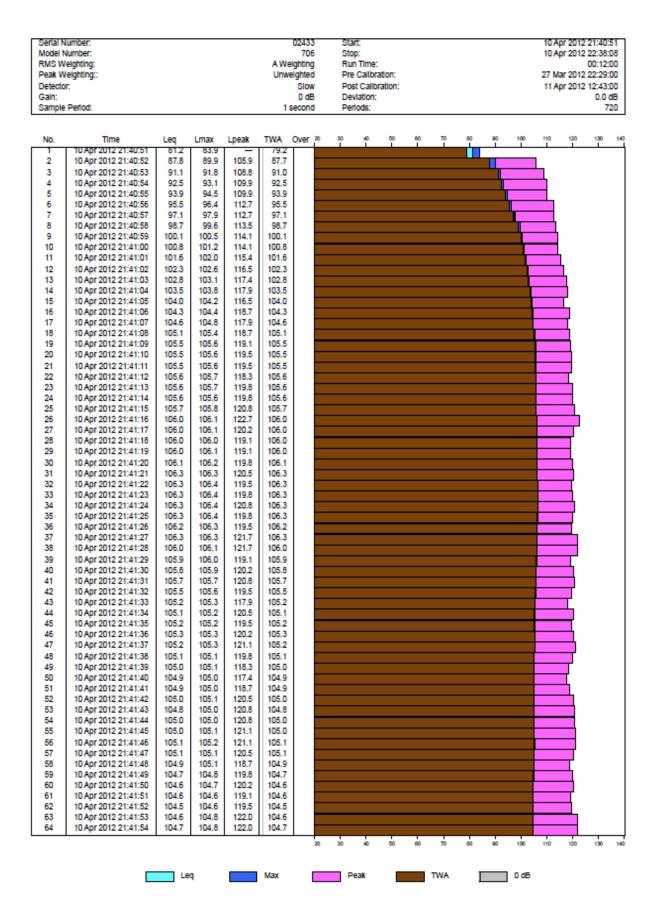


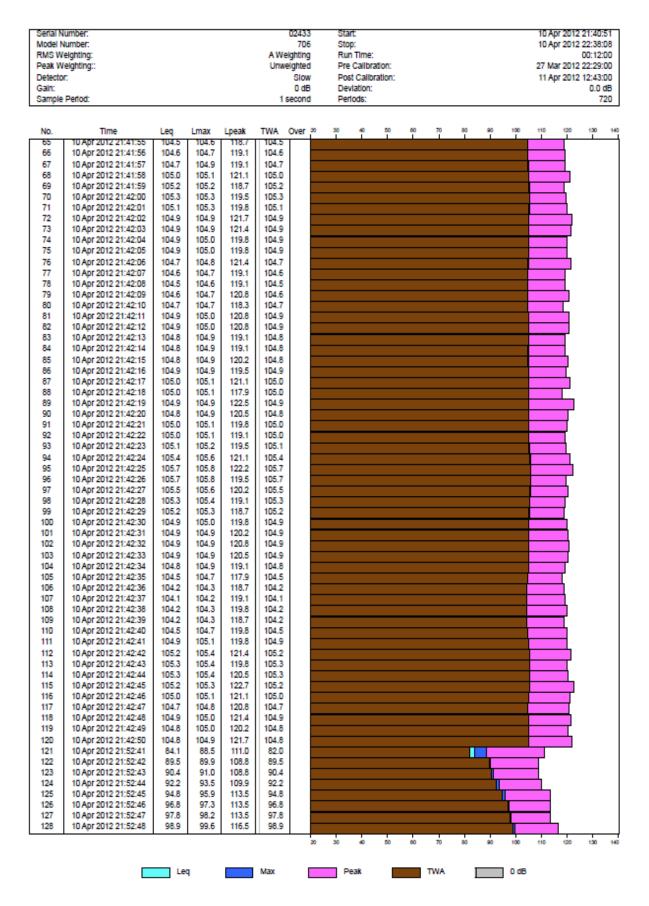
Serial Number:	02433	Start	06 Apr 2012 16:08:10
Model Number:	706	Stop:	06 Apr 2012 17:02:17
RMS Weighting:	A Weighting	Run Time:	00:12:00
Peak Weighting::	Unweighted	Pre Calibration:	27 Mar 2012 22:29:00
Detector:	Slow	Post Calibration:	11 Apr 2012 12:43:00
Gain:	0 dB	Deviation:	0.0 dB
Sample Period:	1 second	Periods:	720
Exchange Rate:	5	Dose:	9.6 %
Threshold:	80.0 dBA	Projected Dose:	385.2 %
Criterion Level:	90.0 dBA	Leq:	100.5 dBA
Criterion Duration:	8.0 hours	TWA:	99.7 dBA
		TWA (8)	73.1 dBA
L10:	103.5 dBA	Lmax	107.7 dBA
L30:	101.0 dBA	Lpeak (max):	121.7 dB
L50:	98.5 dBA	Lep (8)	84.4 dBA
L70:	96.0 dBA	SE:	0.9 Pa ^s hr
L90:	93.0 dBA		

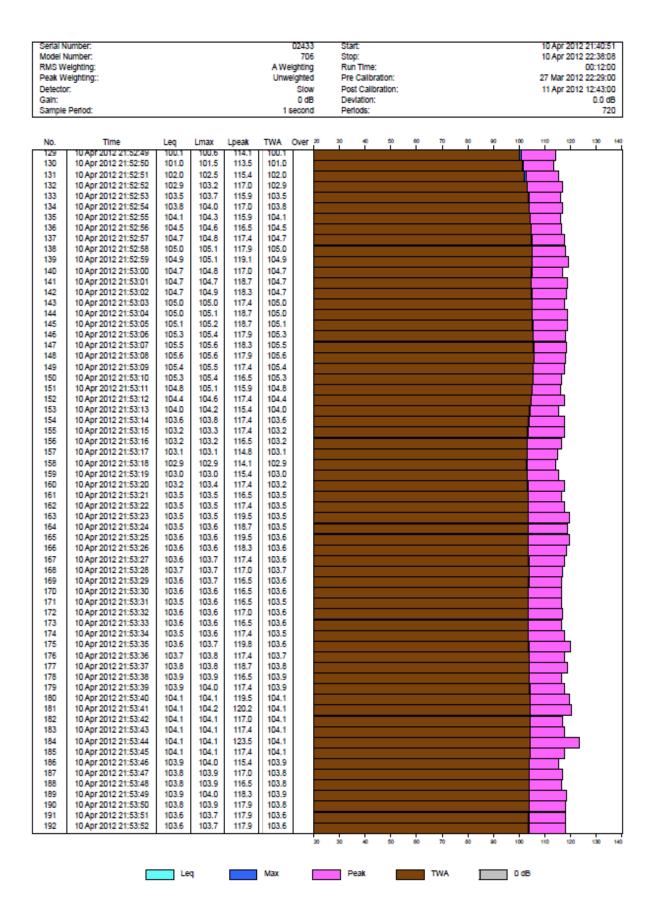


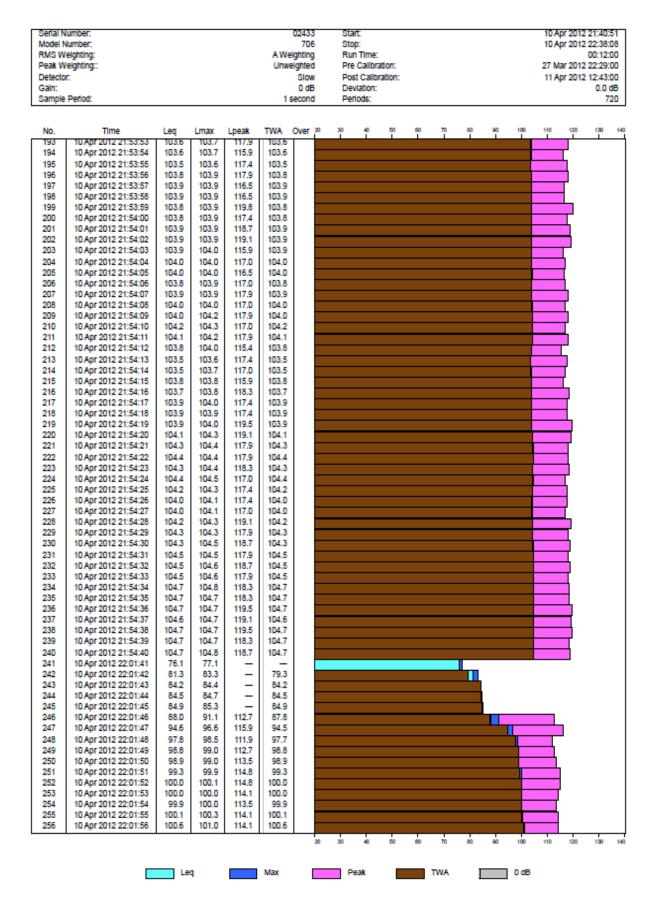
Serial Number: Model Number: RMS Weighting: Peak Weighting:: Detector:	02433 706 A Weighting Unweighted Slow	Start: Stop: Run Time: Pre Calibration: Post Calibration:	10 Apr 2012 22:45:31 10 Apr 2012 23:34:05 00:12:00 27 Mar 2012 22:29:00 11 Apr 2012 12:43:00
Gain: Sample Period:	0 dB 1 second	Deviation: Periods:	0.0 dB 720
Fushmen Date:		5	
Exchange Rate: Threshold:	80.0 dBA	Dose: Projected Dose:	13.6 % 542.3 %
Criterion Level:	90.0 dBA	Leq:	102.5 dBA
Criterion Duration:	8.0 hours	TWA:	102.2 dBA
		TWA (8)	75.6 dBA
L10:	104.0 dBA	Lmax	106.0 dBA
L30:	103.5 dBA	Lpeak (max):	123.2 dB
L50:	102.0 dBA	Lep (8)	85.4 dBA
L70:	101.0 dBA	SE:	1.4 Pathr
L90:	99.0 dBA		

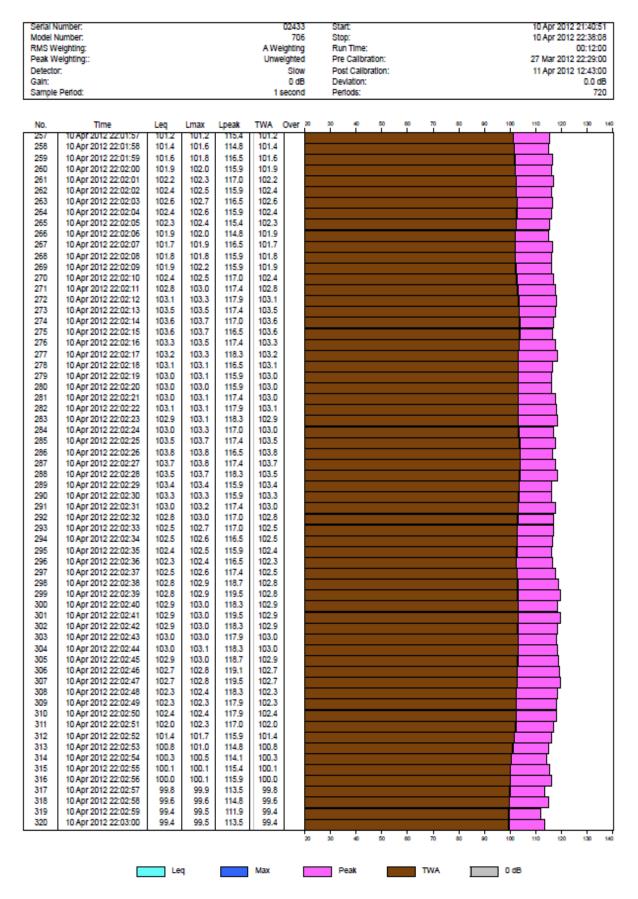


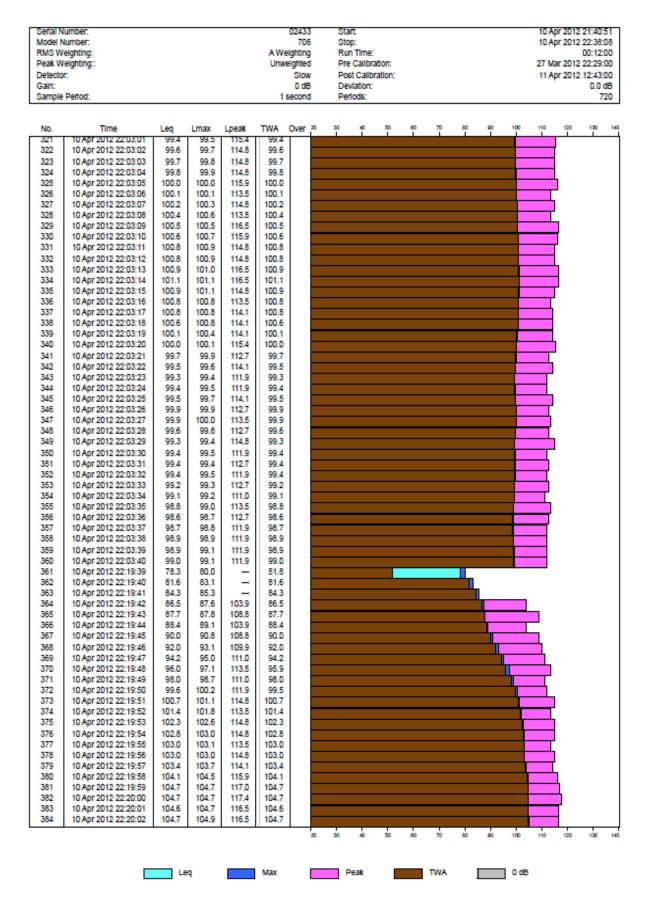


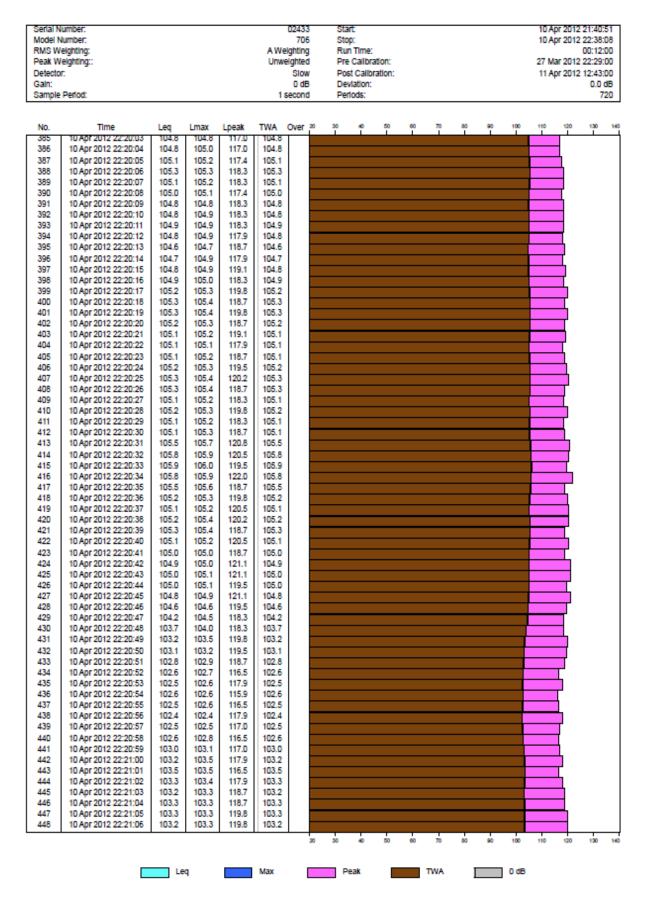


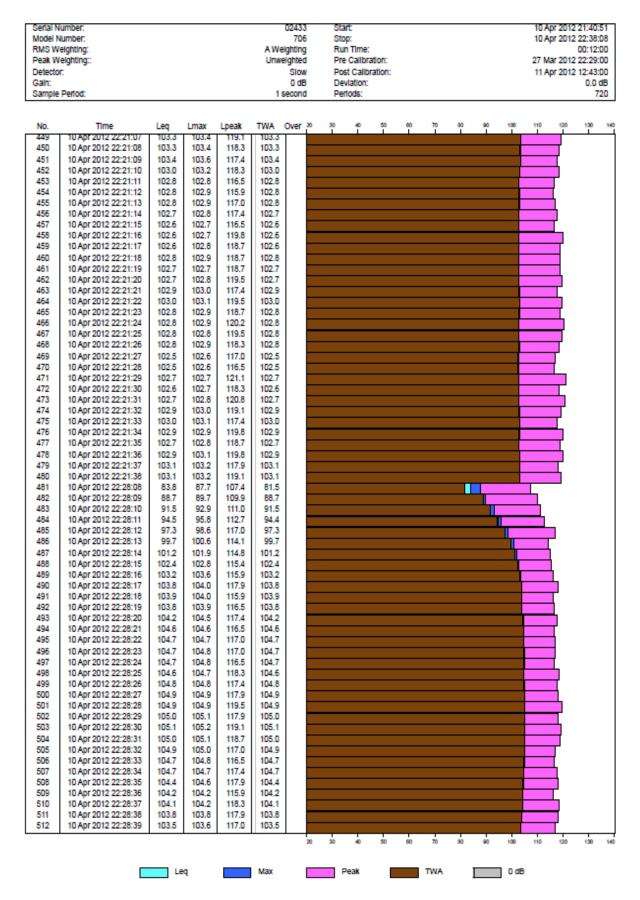


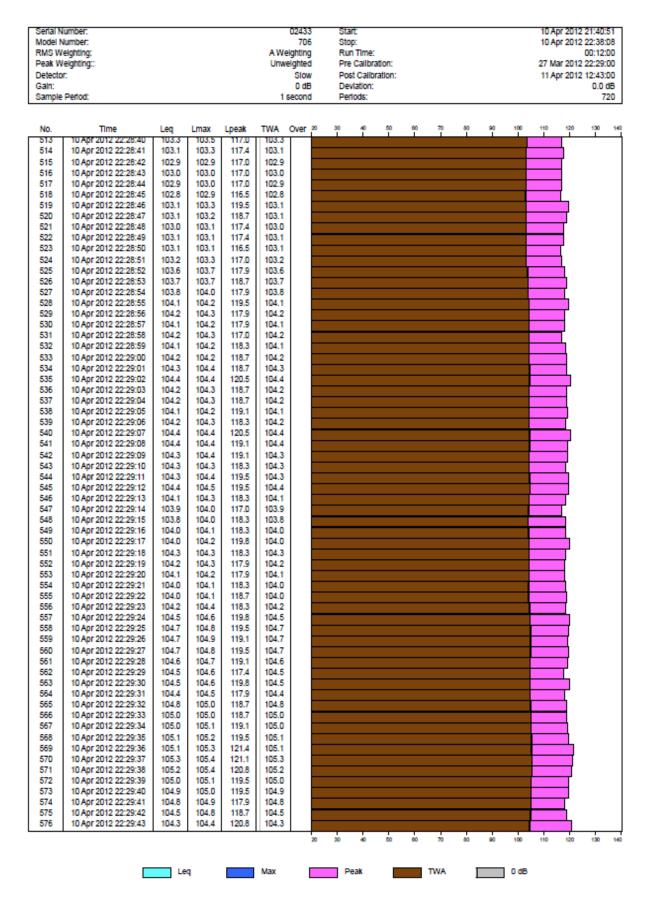


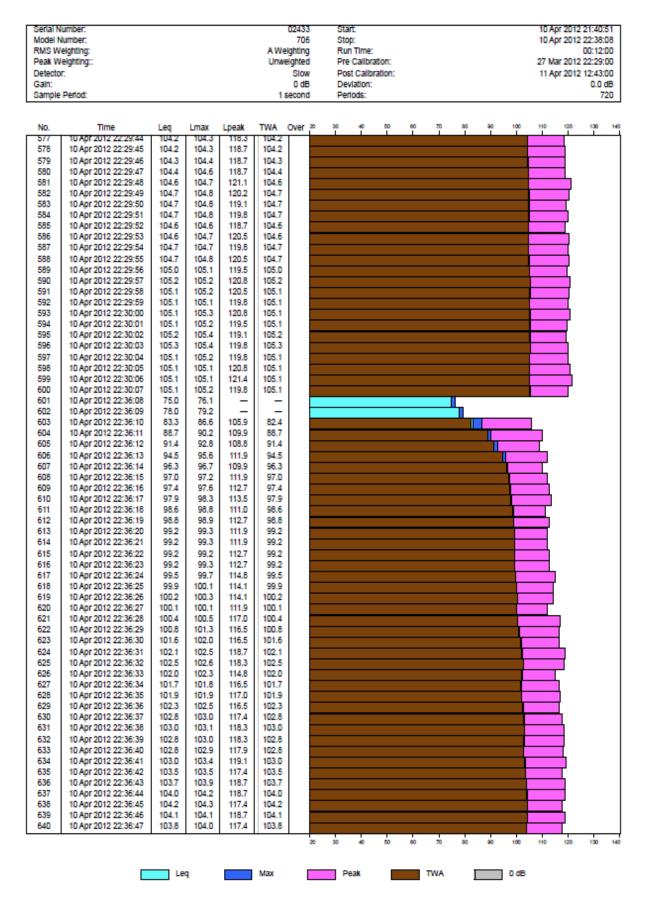




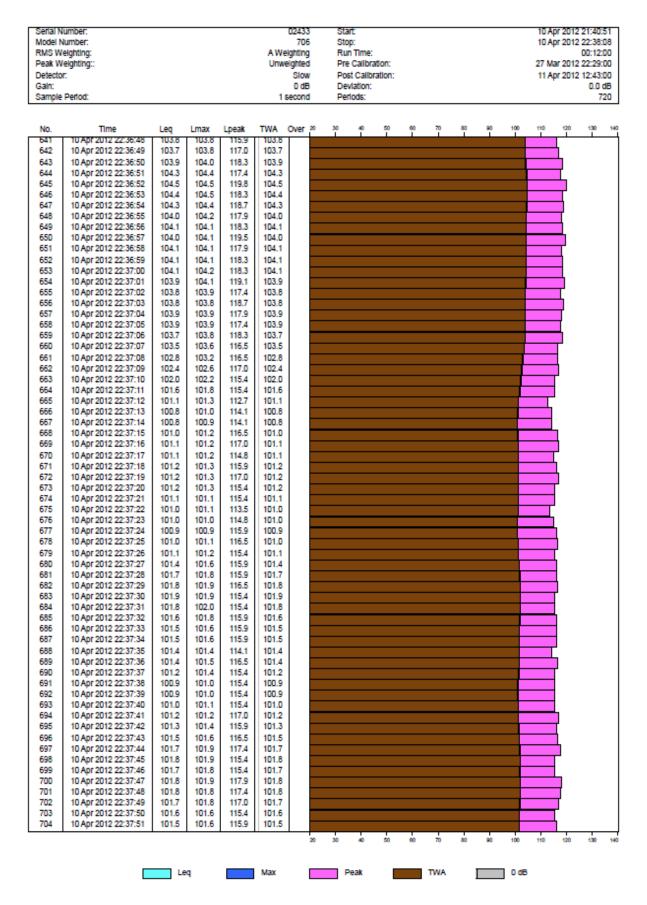








	lumber: leighting: leighting:: r:			Unw	0243 70 elghting elghter Slov 0 de secon	5 9 1 9 8	Pre Pos Dei	p: n Time: e Calibr	ation: ration:						10 Apr 2012 21:40:51 10 Apr 2012 22:38:08 00:12:00 27 Mar 2012 22:29:00 11 Apr 2012 12:43:00 0.0 dB 720				
NO.	Time	Leq	Lmax	Lpeak	TWA	Over	20	30	-	50	60	70	80	90	100	110	120	130	140
705	10 Apr 2012 22:37:52	101.3	101.4	115.9	101.3												_		\neg
706	10 Apr 2012 22:37:53	101.6	101.8	114.1	101.6														
707	10 Apr 2012 22:37:54	101.7	101.8	115.9	101.7												L .		
708 709	10 Apr 2012 22:37:55	101.6	101.7	117.0 117.0	101.6														
710	10 Apr 2012 22:37:56 10 Apr 2012 22:37:57	101.4	101.4	117.0	101.4										_				
710	10 Apr 2012 22:37:57	101.4	101.4	116.5	101.4										_		4		
712	10 Apr 2012 22:37:59	101.4	101.5	114.8	101.4										_				
713	10 Apr 2012 22:38:00	101.3	101.4	115.9	101.3												1		
714	10 Apr 2012 22:38:01	101.3	101.4	114.8	101.3														
715	10 Apr 2012 22:38:02	101.3	101.4	115.9	101.3														
716	10 Apr 2012 22:38:03	101.2	101.2	115.4	101.2														
717	10 Apr 2012 22:38:04	101.3	101.4	115.9	101.3												1		
718	10 Apr 2012 22:38:05	101.5	101.5	117.0	101.4												1		
719	10 Apr 2012 22:38:06	101.3	101.5	115.9	101.3												ľ		
720	10 Apr 2012 22:38:07	101.2	101.3	115.9	101.2				_		_	_	_	_			1		
				-			20	30	4	50	60	70	80	90	100	110	120	130	140



2 06 Aj 3 06 Aj 4 06 Aj 5 06 Aj 6 06 Aj 7 06 Aj 8 06 Aj 9 06 Aj 10 06 Aj 11 06 Aj 11 06 Aj 13 06 Aj 14 06 Aj 15 06 Aj 16 06 Aj 16 06 Aj 17 06 Aj	Time pr 2012 16:08:10 pr 2012 16:08:12 pr 2012 16:08:13 pr 2012 16:08:14 pr 2012 16:08:16 pr 2012 16:08:16 pr 2012 16:08:17 pr 2012 16:08:18 pr 2012 16:08:19 pr 2012 16:08:19	Leq 85.8 88.2 90.6 92.5 94.1 95.3 96.5	Lmax 86.7 89.5 91.5 93.3	Lpeak 105.9	TWA 85.8	Over	20			Start: Stop: Run Time: Pre Calibration: Post Calibration: Devlation: Periods:									
2 06 Aj 3 06 Aj 4 06 Aj 5 06 Aj 6 06 Aj 7 06 Aj 8 06 Aj 9 06 Aj 10 06 Aj 11 06 Aj 11 06 Aj 13 06 Aj 14 06 Aj 15 06 Aj 16 06 Aj 16 06 Aj 17 06 Aj	pr 2012 16:08:11 pr 2012 16:08:12 pr 2012 16:08:13 pr 2012 16:08:14 pr 2012 16:08:16 pr 2012 16:08:16 pr 2012 16:08:17 pr 2012 16:08:17 pr 2012 16:08:18 pr 2012 16:08:19	88.2 90.6 92.5 94.1 95.3 96.5	89.5 91.5	105.9	85.8			30	-	50	60	70	80	90	100	110	120	130	140
3 06 A 4 06 A 5 06 A 6 06 A 7 06 A 9 06 A 9 06 A 10 06 A 11 06 A 11 06 A 12 06 A 13 06 A 14 06 A 15 06 A 16 06 A 16 06 A 17 06 A 17 06 A 10 06 A 11 06 A 11 06 A 12 06 A 13 06 A 13 06 A 14 06 A 15 06 A 15 06 A 16 06 A 17 06 A 10 06 A 10 06 A 10 06 A 11 06 A 12 06 A 13 06 A 14 06 A 15 06 A 15 06 A 16 06 A 16 06 A 17 06 A 16 06 A 17 06 A 17 06 A 16 06 A 17 06 A 16 06 A 17 06 A 16 06 A 17 06 A 16 06 A 17 06 A 17 06 A 16 06 A 17 06 A 17 06 A 16 06 A 17 06 A 16 06 A 16 06 A 17 06 A 16 06 A 17 06 A 16 06 A 16 06 A 16 06 A 16 06 A 16 06 A 17 06 A 16 06 A 16 06 A 16 06 A 16 06 A 16 06 A 17 06 A 16 06 A 1	pr 2012 16:08:12 pr 2012 16:08:13 pr 2012 16:08:14 pr 2012 16:08:15 pr 2012 16:08:16 pr 2012 16:08:16 pr 2012 16:08:18 pr 2012 16:08:19	90.6 92.5 94.1 95.3 96.5	91.5	105.9															
4 06 A 5 06 A 6 06 A 7 06 A 9 06 A 9 06 A 10 06 A 11 06 A 12 06 A 13 06 A 14 06 A 14 06 A 15 06 A 16 06 A 17 06 A 17 06 A	pr 2012 16:08:13 pr 2012 16:08:14 pr 2012 16:08:15 pr 2012 16:08:15 pr 2012 16:08:16 pr 2012 16:08:17 pr 2012 16:08:18 pr 2012 16:08:19	92.5 94.1 95.3 96.5			88.2											1			
5 06 A 6 06 A 7 06 A 9 06 A 9 06 A 10 06 A 11 06 A 12 06 A 13 06 A 14 06 A 15 06 A 16 06 A 16 06 A 17 06 A 17 06 A	pr 2012 16:08:14 pr 2012 16:08:15 pr 2012 16:08:16 pr 2012 16:08:17 pr 2012 16:08:17 pr 2012 16:08:18 pr 2012 16:08:19	94.1 95.3 96.5	50.0	107.4	90.6 92.5									┛					
6 06 A; 7 06 A; 8 06 A; 9 06 A; 10 06 A; 11 06 A; 12 06 A; 13 06 A; 13 06 A; 14 06 A; 15 06 A; 16 06 A; 17 06 A;	pr 2012 16:08:15 pr 2012 16:08:16 pr 2012 16:08:17 pr 2012 16:08:18 pr 2012 16:08:19	95.3 96.5	94.7	112.7	94.1														
8 06 A 9 06 A 10 06 A 11 06 A 12 06 A 13 06 A 13 06 A 15 06 A 16 06 A 17 06 A	pr 2012 16:08:17 pr 2012 16:08:18 pr 2012 16:08:19		95.8	114.8	95.3											<u> </u>			
9 06 A 10 06 A 11 06 A 12 06 A 13 06 A 14 06 A 15 06 A 16 06 A 16 06 A 17 06 A	pr 2012 16:08:18 pr 2012 16:08:19		97.3	112.7	96.5														
10 06 Ay 11 06 Ay 12 06 Ay 12 06 Ay 13 06 Ay 14 06 Ay 15 06 Ay 15 06 Ay 16 06 Ay 17 06 Ay	pr 2012 16:08:19	98.0	98.5	114.1 114.1	98.0 99.0														
11 06 A 12 06 A 13 06 A 14 06 A 15 06 A 16 06 A 17 06 A		99.0 99.8	99.4 100.1	114.1	99.8			_	_						_				
13 06 Ap 14 06 Ap 15 06 Ap 16 06 Ap 17 06 Ap	or 2012 16:08:20	100.4	100.8	115.4	100.4														
14 06 Ap 15 06 Ap 16 06 Ap 17 06 Ap	pr 2012 16:08:21	101.1	101.3	117.0	101.1														
15 06 Ap 16 06 Ap 17 06 Ap	pr 2012 16:08:22	101.4	101.6	117.0	101.4												1		
16 06 Ap 17 06 Ap	pr 2012 16:08:23	101.8	101.9	117.9	101.8														
17 06 Ap	pr 2012 16:08:24 pr 2012 16:08:25	101.9	101.9	117.0 115.4	101.9														
	or 2012 16:08:26	101.5	102.0	115.9	102.0												1		
10 00/0	or 2012 16:08:27	102.0	102.2	115.4	102.0														
	pr 2012 16:08:28	102.2	102.2	115.4	102.2														
	pr 2012 16:08:29	102.3	102.5	115.9	102.3												1		
	pr 2012 16:08:30 pr 2012 16:08:31	102.5	102.5 102.5	116.5 115.4	102.5												l I		
	or 2012 16:08:32	102.4	102.5	117.0	102.4												•		
	or 2012 16:08:33	102.9	103.1	115.9	102.9												1		
	pr 2012 16:08:34	103.1	103.2	115.9	103.1														
	pr 2012 16:08:35	103.2	103.2	116.5	103.2														
	pr 2012 16:08:36 pr 2012 16:08:37	103.2	103.3	116.5 116.5	103.2												4		
	or 2012 16:08:38	103.1	103.2	115.9	102.9												ļ		
	or 2012 16:08:39	102.8	102.9	115.9	102.8														
	or 2012 16:08:40	102.4	102.5	116.5	102.4												1		
	pr 2012 16:08:41	102.2	102.3	114.1	102.2														
	pr 2012 16:08:42	102.0	102.1	115.4 115.9	102.0												4		
	pr 2012 16:08:43 pr 2012 16:08:44	101.7	102.0	115.9	101.7												4		
	or 2012 16:08:45	101.5	101.7	117.0	101.5												4		
	or 2012 16:08:46	101.7	102.1	117.9	101.7												h		
	pr 2012 16:08:47	102.4	102.7	116.5	102.4												Г		
	pr 2012 16:08:48	102.6	102.7	116.5	102.6												1		
	pr 2012 16:08:49 pr 2012 16:08:50	102.9	102.9	117.0 114.8	102.9														
	or 2012 16:08:51	102.1	102.5	115.4	102.1														
	or 2012 16:08:52	101.2	101.5	114.8	101.2														
	pr 2012 16:08:53	101.8	102.4	117.4	101.8														
	pr 2012 16:08:54	102.6	102.7	117.4	102.6														
	pr 2012 16:08:55 pr 2012 16:08:56	102.7	102.8	115.9 117.0	102.7												L		
	or 2012 16:08:57	102.9	102.9	116.5	102.9												4		
	or 2012 16:08:58	102.9	102.9	117.4	102.9												4		
	pr 2012 16:08:59	103.0	103.1	117.9	103.0												1		
	pr 2012 16:09:00	103.1	103.1	117.4	103.1														
	pr 2012 16:09:01 pr 2012 16:09:02	103.1 103.2	103.2 103.3	116.5 116.5	103.1 103.2												4		
	or 2012 16:09:02	103.2	103.3	110.5	103.2												4		
	pr 2012 16:09:04	103.3	103.4	117.4	103.3												-		
	pr 2012 16:09:05	103.2	103.3	116.5	103.2												ſ		
	pr 2012 16:09:06	103.2	103.3	118.3	103.2														
	pr 2012 16:09:07	103.2	103.3	117.0	103.2												Į.		
	pr 2012 16:09:08 pr 2012 16:09:09	103.3 103.0	103.3 103.2	115.9 115.9	103.3														
	pr 2012 16:09:10	102.7	103.2	116.5	102.7												1		
62 06 Ap	pr 2012 16:09:11	102.5	102.7	116.5	102.5											_	1		- 1
		102.4	102.5	116.5	102.4	1													1
64 06 Ap	pr 2012 16:09:12 pr 2012 16:09:13	102.3	102.5	114.8	102.4	1											1		