

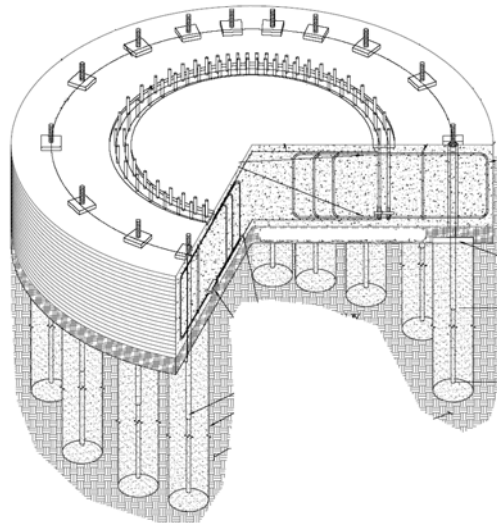


**ENGINEERING DESIGN AND ANALYSIS OF THE  
PATRICK AND HENDERSON ROCK ANCHOR FOUNDATION  
Stetson Mountain Wind Project, Danforth, Maine**

**38 – GE Wind 1.5sle 1.5MW on 80-m Hub Height Towers**

14 – 40 to 50-foot Rock Anchors on 20-foot Ring with 24-foot Diameter Cap  
using Williams 2-1/2-in diameter Grade 150 ksi Anchors or Equal

September 16, 2007



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September 17, 2007

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OF THE PATRICK AND HENDERSON ROCK ANCHOR FOUNDATION  
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The Patrick and Henderson (P&H) rock anchor foundation is a proprietary (patent pending) foundation used to support wind turbines on monopole steel towers. This narrative presents the engineering design and analyses of the structural and geotechnical strength, stability and stiffness of the rock anchor foundation. The internal structural analysis of the cap is presented from Patrick and Henderson, Inc. *This document is for limited distribution as it contains confidential and proprietary information subject to confidentiality restrictions. This document and all related calculations and correspondence are the intellectual property of Earth Systems Southwest and is copyrighted with all rights reserved.*

### **Description**

The P&H rock anchor foundation consists of a 5-foot thick, 24-foot diameter, reinforced concrete mat (cap) supported by 14 – mostly 40 feet deep (minimum), rock anchors aligned within a 20-foot diameter circle. For turbine sites with rock at depths of 10 feet or less from top of the foundation, 40 foot long rock anchors are adequate and the mud mat should extend to the top of rock. **For Turbine Sites T4 and T33 with the depth to rock at about 12.9 and 9.8 feet below expected mud mat elevations, respectively, the anchor lengths should be extended to 50 feet.** *The rock anchor foundation size and anchor lengths are based in part on the successful completion of the Mars Hills, Maine project using the same foundation system with the same wind turbine and similar geologic conditions.*

Williams 2 ½ inch diameter rods are used within the rock anchors. The number and length of anchors are based on the magnitude of applied loads at the top of the foundation as well as the soil and rock conditions. The rock anchors are installed by drilling a shaft and filling the shaft with the anchor rod and grout. Post-tensioning is used to develop an internal tension force in each anchor. The intention is to have enough tension in the rod to reduce foundation deflections from normal wind loading.

The vertical loads are transferred through a base plate connection at the top of the circular plate with a double bolt circle (inner and outer bolt rings). Axial loads are transferred along the circular plate circumference via the concrete slab to the subgrade and circumferential rock anchors. The rock anchors then transfer the axial loads into the soil or rock through skin friction. Lateral loads are resisted by base shear friction at the bottom of the cap and lateral pressures along the anchors.

Overturning moments are resisted by skin friction along the length of the rock anchors. The anchors are pretensioned to create a compression stress on the subgrade beneath the cap. The prestress effect resists the overturning moment by creating a built-in clamping force to keep the cap in place. The higher the pretensioning of the anchors results in a higher moment capacity of the foundation. The optimum point of prestress is a function of the resisting moment versus demand and factor of safety. The prestress in the anchor is calculated to cover the load imposed by maximum wind forces given by the turbine manufacturer. Therefore, normal operational loads will not produce stress reversals in the subgrade/concrete interface or anchor. There is no substantial cyclic degradation of the soil or rock.

## **Design Criteria and Assumptions**

### ***Referenced Code, Standards, and Industry Recommended Practices***

The design of P&H rock anchor foundations uses a combination of national and international codes and standards. In general, the design of wind turbine foundations should be in accordance with the referenced national standards, codes of practice, legislation, licenses and consents applicable to the design and construction of the wind turbine foundation at each particular wind turbine location. In addition, the design of the wind turbine foundations should also be in accordance with the requirements of the International Electrotechnical Committee (“IEC”). The structural design loads and factors within the national codes of the United States (e.g. the International Building Code, the Uniform Building Code and ASCE-7, *Minimum Design Loads for Buildings & Structures*) were developed specifically for building structures. These codes consider wind turbines as non-building structures. The IEC developed the IEC 61400-1 standard specifically for wind turbines through extensive research.

The governing code for most jurisdictions in the United States is the International Building Code (IBC). Structural loads and factors for the foundation are derived in combination with the IEC 61400-1, “*Wind Turbine Generator Systems, Part 1: Safety Requirements*” and from Chapter 16 of the IBC that has its basis from the ASCE 7 standard. Chapter 18 of the IBC addresses foundation design in a general manner, but without definitive design equations applicable for the geotechnical design of the P&H anchor foundation. The structural design is in accordance with Chapter 19 of the IBC that has its basis on the requirements of the American Concrete Institute’s “*Building Code Requirements for Structural Concrete*” (ACI 318). The design of the anchor themselves are based on the Post-Tensioning Institute, “*Recommendations for Prestressed Rock and Soil Anchors*”.

Design manuals from the US Departments of Defense and Transportation and supplemented by peer reviewed technical journal articles from the American Society of Civil Engineers (ASCE) provide design guidance for the analyses as cited in the references. The “*Guidelines for Design of Wind Turbines*” published by DNV and Risø National Laboratory in Denmark provides general design guidance.

The following codes, standards and industry recommended practices are the main references used in the calculations and this document:

1. American Concrete Institute Building Code Requirements for Reinforced Concrete (ACI 318)
2. American Concrete Institute Suggested Design and Analysis and Design Procedures for Combined Footings and Mats (ACI 336.2)
3. Federal Highway Administration Geotechnical Engineering Circular No. 4, Ground Anchors and Anchored Systems (FHWA-IF-99-015)
4. International Building Code, 2003 edition

## 5. Post-Tensioning Institute Recommendations for Prestressed Rock and Soil Anchors (PTI)

In addition to the codes and geotechnical references noted above, the design of wind turbine foundations for a particular site should also be in accordance with a site-specific geotechnical report. The wind loads applied to each particular wind turbine foundation should be in accordance with the site-specific wind load data provided by the wind turbine manufacturer.

**Design Criteria**

Design Criteria: The P&H rock anchor foundation is designed for use with a GE Wind 1.5sls MTS 1.5 megawatt (“MW”) wind turbine generator (“WTG”) with an 77-meter (“m”) rotor diameter and a 80-m hub height. This WTG is designed by Gamesa in accordance with the International Electrotechnical Commission (“IEC”) requirements for the IEC Class II<sub>A</sub> wind load regime indicated in Table 1. Based upon these parameters, GE Energy provided the wind loads in a document entitled “*Foundation Data for Wind Turbine Generator Systems, GE 1.5sls MTS, 61.4-85m HH, IEC TCIIa with reduced gust ( $V_{e50} = 55m/s$ ), LM 37.3P2, 50&60 Hz, GE 37c 60 Hz, dated March 23, 2006*”. In this document, GE Energy provided foundation loads for the extreme wind for seven different cases, as shown in Table 2 (Earthquake loading not shown, does not govern for this site). Additionally, GE Energy provided minimum stiffness requirements for rotation and translation, as shown in Table 3.

**Table 1**  
**IEC IIA Wind Load Parameters**

Annual Average Wind Speed (m/s)	8.5
Annual Average Air Density (kg/m <sup>3</sup> )	1.225
Reduced Extreme Gust (3 sec, m/s)	55 (per GE)
Average Turbulence Intensity (I15)	0.16

Design Loads: The most relevant design loads provided in the load document for the GE Wind 1.5 MW turbines for the project are summarized in the table below.

**Table 2**  
**GE Wind 1.5sle 1.5 MW, 77-m rotor on 80 m HH tower, Foundation Loads**

Load Case Description	Blade Type	Design Load Case DLC	Load Factor LF	Vertical Fz KN	Horizontal Fr KN	Moment Mr KN-m	% M extreme
4.1 Extreme Load	GE37c	6.1j	1.00	-1844.5	523.4	34907.7	100%
4.2 Lift-off	LM37.3p2	1.0	1.00	-1958.9	219	17596.2	50%
4.3 Tilting	GE37c	6.1j	1.00	-1844.5	522.6	34907.7	100%
4.4 Sliding	GE37c	6.1j	1.00	-1843.4	523.4	34863.1	100%
4.5 Shear Failure	GE37c	6.1j	1.35	-2490	705.6	47125.1	135%
4.6 Tension Loading	LM37.3p2	1.1	1.00	-1955.4	327.4	24238.1	69%

**Table 3**  
**GE Wind Minimum Stiffness Requirements**

<u>Description</u>	
Minimum stiffness for rotation about a horizontal axis	30 GN-m/radian

### **Geotechnical Parameters**

The geotechnical borings by SW Cole and the geophysical investigation by Hager Richter Geosciences, Inc. are the basis for the geotechnical properties of the rock selected for the analyses. Representative rock core borings have been made to characterize each of the 38 turbines sites. The subsurface profile generally consists of dense glacial till of sands and gravels to variable depths to up to about 21.6 foot depth, but typically less than 6 foot depth. Beneath the soil overburden, generally lies variably weathered, fractured, meta siltstone (pelite) and volcanic rock (musdstone). The compressive strength of intact rock cores ranges from 2360 to 28090 psi, but generally above 3120 psi, and averaging about 6200 psi.

### **Design Methodology**

Serviceability limit state (SLS) design is based on normal and fatigue wind loading events. Ultimate limit state (ULS) design is based on extreme wind loading events.

### **Load Cases**

Based upon the information in ACI 318 and IEC 61400-1, the following load cases are considered in the design of the P&H rock anchor foundation:

#### Case 1 - Foundation Stability

Overturning and bearing strength for the foundation under unfactored Extreme wind loads

#### Case 2 - Foundation Strength

Strength of the foundation under factored Extreme wind loads

### **Partial Load Factors**

The design of the foundation should consider the applicable partial load factors for ultimate limit state and serviceability design. For ultimate limit state design using Extreme and Abnormal conditions, the partial load factors in Table 4 are used. For the serviceability limit state design, the partial load factors in Table 5 are used.

**Table 4**  
**Load Factors for Ultimate Limit State Design**  
**Extreme Wind Loading - DLC 6.2**

<u>Parameter</u>	<u>Case 1</u> <u>Stability</u>	<u>Case 2</u> <u>Strength</u>
Wind Turbine Load <sup>(1)</sup>	1.0	1.35
Dead Load (unfavorable) <sup>(2)</sup>	1.0	1.1/1.35
Dead Load (favorable)	0.9	0.9
Hydrostatic Pressure (unfavorable)	1.0	1.5
Hydrostatic Pressure (favorable)	0.0	0.0
Minimum Permissible Factor of Safety	1.5	n/a

- (1) For seismic loading, the “wind turbine load” consists of the required seismic load applied to the wind turbine structure combined with the Normal Operating load condition. Both sets of loads are factored by the same partial load factor shown in the table.
- (2) For manufactured wind turbine components, and materials that exhibit ductile behavior (e.g. anchor bolts), a partial safety factor of 1.1 may be used. For all other elements of the foundation system, a partial safety factor of 1.35 is used.

**Table 5**  
**Load Factors for Serviceability and Fatigue Limit State Design**  
**Normal Operating Loads - DLC 1.1**

<u>Parameter</u>	<u>Serviceability</u> <u>Checks</u>	<u>Case 3</u> <u>Soil/Rock</u> <u>Stability</u>	<u>Fatigue Life</u> <u>Assessment</u>
Annual Turbine Load	1.0	1.0	n/a
Equivalent (20-year) Fatigue Load	n/a	n/a	1.0
Dead Load (unfavorable)	1.0	1.0	1.0
Dead Load (favorable)	0.9	0.9	1.0
Hydrostatic Pressure (unfavorable)	1.0	0.0	1.0
Hydrostatic Pressure (favorable)	0.0	0.0	0.0
Limiting Design Value	Per Table 6	3.0	0.33

### Serviceability Criteria

The design of foundations considers the action of the foundation under the applicable serviceability loads and estimated foundation movements and settlements. Additionally, the structural calculations demonstrate that the foundation will remain serviceable after the application of the Extreme and Abnormal loadings. Allowable movements and settlements under serviceability loadings are indicated below.

**Table 6**  
**Serviceability Criteria**

<u>Average Settlement</u> <u>(max mm)</u>	<u>Rotation</u> <u>(max radians)</u>	<u>Minimum Rotational Stiffness</u> <u>(GN-m/radian)</u>
25	(0.001) Operational load (0.002) E-stop load (0.004) Extreme load	30

### Rock anchor Foundation Components

The P&H rock anchor foundation consists of a circular reinforced concrete cap supported by post-tensioned (also called prestressed) rock anchors. The cap measures 24-feet in diameter, with a thickness of five feet. The 14 rock anchors that support the cap are equally spaced on a 20-foot diameter circle. Each rock anchor consists of a 40-foot deep, 5-inch diameter grouted hole that is post-tensioned using 2½-inch diameter, 150 kips per square inch (ksi) anchor rods that extend full depth.

Table 7 summarizes that key design parameters and specification shown on the drawings and the calculations.

**Table 7**  
**Summary – Design and Specification**

<u>Item &amp; Parameter</u>	<u>Specification</u>	<u>No. &amp; Dimensions</u>
Cap - Concrete Strength	6,000 psi	24 foot diameter x 5 feet thick
Rock Anchor - Grout Strength	3,000 psi	14 -5 inch diameter x 40 feet long (mimimum)
Tower Anchor Bolt - Strength	Grade 150	112 - 1.375 inch diameter x 66 inch long
Tower Anchor Bolt - Prestress	120 kips	
Rock Anchor Bolt - Strength	Grade 150	
Rock Anchor Bolt - Prestress	435 kips	(minimum required 319 kips)
Rock Anchor Base Plate - Strength	50 ksi	14 - 12 inch square x 2.5 inches thick
Embedment Material	36 ksi	
Grout beneath Flange - Strength	10,000 psi	
CMP Material	A-929 or A-444	28 foot diameter x 5 feet long
Rock anchor - Free Stress Length = 10 feet		
Rock anchor - Bonded Length = 28.5 feet (minimum in rock)		

### QA/QC Load Testing

A Patrick & Henderson, Inc. representative shall be present during anchor installation to verify, and to document via the checklists provided by P&H, that the conditions encountered are consistent with the conditions as indicated in the design calculations. All anchors will be tested in accordance with PTI recommendations for prestressed anchors. All anchors will be proof-tested to 133% of the design post-tensioned load before lockoff. At least one anchor per foundation will be performance tested with load and reload cycles.

The apparent free length of the anchor,  $L_a$ , may be calculated using the following equation:

$$L_a = \delta E A_t / P$$



Where,  $A_t$  is the cross sectional area of the prestressing steel,  $E_s$  is the Young's modulus of the prestressing steel,  $\delta$  is the elastic movement at the test load, and  $P$  is equal to the test load minus the alignment load.

The minimum apparent free length is defined as the jack length plus 80 percent of the design unbonded length. The maximum value of apparent free length is restricted to elastic movements of 100 percent of the free length plus 50 percent of the bond length plus the jack length.

## DESIGN METHODS

### Part A - Geotechnical Design and Analysis (external force analysis)

Unless otherwise noted, the geotechnical design follows the recommendations of the site-specific geotechnical report. Unless otherwise noted, bearing capacity and lateral resistance calculations are in accordance with the parameters given with the recommendations of the site-specific geotechnical report.

Moment Capacity: The analysis must fundamentally demonstrate the rock anchor foundation is reasonably safe against complete upset by overturning. A global safety factor of at least 1.5 should be used in the ultimate limit-state (ULS) analysis for extreme wind loads as discussed in detail below.

Dynamic Stiffness: Dynamic analyses of the foundation are conducted to verify that the stiffness of the combined foundation-soil system is such that resonance and excessive vibrations (that potentially could lead to premature fatigue of the tower) are minimized. The stiffness of the foundation must avoid the critical first and third vibrational mode frequencies to limit dynamic amplification or resonance.

Rotational stiffness plays an important role in the foundation performance. Rotational stiffness typically expressed in units of GN-m/radian is defined as Moment ( $M$ ) divided by rotation ( $\theta$ ) expressed in radians. To verify that the foundation-soil system remains outside of the critical frequency range, the rotational stiffness is evaluated and checked to whether that it meets the rotational stiffness requirements of the turbine manufacturer. It must be sufficiently large so as not to significantly alter the total system frequency of tower/nacelle/foundation to change the turbine and tower manufacturers fatigue load assumptions. Evaluation of the stiffness of the P&H anchor foundation system is based upon the vertical dead loads (including sustained post-tension force) and the maximum mean overturning moment due to operational loads specified by the wind turbine manufacturer's fatigue load analysis.

Design Post-Tension Load: The cap is structurally connected at the top bearing plate to the anchors. These high strength steel anchors that are post tensioned to maintain contact between the cap and anchor during most of the loads. The post-tension load is selected so that the cap and the anchors remain in contact for the normally encountered operational loads, and to minimize the deflections during a design basis event. It is not required for structural adequacy, to fully tighten the anchors to the value of tension that would occur due to the extreme design load. An anchor tension somewhat less than the maximum tension may be used.

After the post-tension load is determined, the maximum tension in the anchor is estimated. The anchor capacity is checked against pull-out with respect to the estimated maximum tension load. The mode in which the rock anchor resists the tension or compression load is via bond stress. Bond stress is the skin friction between the grout and the surrounding rock. The foundation is anchored (bonded) at depths of 10 to 44 feet. The foundation is checked against the design overturning moment similar to a spread foundation on rock except the rock anchors induced a tremendous vertical restraining force that keeps the foundation cap clamped down.

**Computations**

Anchor Structural Capacity: The structural capacity of the rock anchor is limited by both the allowable tension load on the anchor taken as 60% of ultimate strength of the anchor rod,  $0.6F_u$ .

Anchor Pullout (Geotechnical) Capacity: The anchor pullout capacity is a function of bond stress between the rock and the grout. Bond stress is dependent on the soil/rock type, condition, and the method of grouting. The ultimate geotechnical uplift capacity of the rock anchors for is given by the summation of perimeter skin bond stress:

$$Q_p = \sum_{z=0}^{z=l} f_s * \pi * d * \Delta z$$

where,

$f_s$  = ultimate bond stress, see discussion below

$d$  = diameter of anchor or anchor group as shown below, used nominal diameter.

$\Delta z$  = increment of depth along anchor bonded length

$l$  = bonded length of anchor below cap

Based on the in-situ condition of the rock, the following ultimate bond stress value was assigned to the rock:

<u>Rock Type &amp; Condition</u>	<u>Ultimate Bond Stress</u>
Fractured weathered rock	150 psi (1 MPa)

This value is based on PTI recommendations for rock type, and our experience with the successfully completed Mars Hill Maine project with similar geologic conditions and is only about 5% of the lower bound of intact compressive rock core strengths.

The length of anchor within the soil that will resist the design tension load is called the bond length. After the bond length is determined, a second calculation is performed to check that there is enough soil/rock mass above the bond length to resist the design loads. The rock mass that is considered for resistance of uplift is a cylinder and cone extending from the perimeter of the anchor group.

**Moment Capacity and Overturning Stability**

P&H Design Methodology: The preliminary design of the P&H rock anchor foundation is based on a simple premise to specify the anchor rod size, number of anchors, and anchor circle diameter so that the post-tensioning force (when set near the allowable structural capacity of the rod) is greater than the force couple created by the characteristic (unfactored), extreme wind overturning moment. The force couple is given as:

$$M = nPD/4,$$

Where,

$P$  = maximum tension or compressive force in the direction of loading,

$M$  = moment,

$n$  = number of anchors = 14,  $D$  = effective diameter of anchor circle = 20 feet.

Hence the post-tension load,  $P$  is set near or above  $4M_{(extreme)}/nD - 0.9DL = 319$  kips,

where  $DL$  = dead weight of foundation cap and turbine

Typically the specified Ps near  $0.6F_u = 435$  kips, to allow for at least 10% creep losses.

In that all anchors are proof-tested to 133% of the post-tensioned, they are in effect tested to the factored extreme wind load.

Overturning Stability with Post-Tension Load on Anchors: The overturning stability is evaluated similar to a spread foundation, whereby the load eccentricity ( $e$ ) is defined by  $e = M/V$ , where  $M$  is the net overturning moment at the base of the cap and  $V$  is the sum of vertical forces (including the post-tensioned anchor force). In accordance with ACI 336.2-R88, the Stability Ratio (SR) (aka safety factor, FS) is evaluated to verify a SR of least 1.5 using the unfactored extreme wind load considering the  $R_{vmin}$  defined as the least resultant of all forces (including preload) acting perpendicular to base area under any condition of loading simultaneous with the overturning moment. The stability ratio (SR) or factor of safety (FS) against overturning is taken as  $FS = e_{max} / e$ . The net driving moment  $M_D$  is the sum of driving moments at the base (computed as  $F_x \cdot (e_z + h)$  – (any reliable passive and vertical side friction moment along the cap, generally neglected). The resisting (restoring) moments  $M_R$  is the base (or anchor group) reaction ( $R$ ) at maximum allowable eccentricity,  $e_{max}$ , taken at  $e_{max} = 0.5B$  for solid rock (at point of over toppling). The factor of safety becomes There are two stability checks considered as follows:

#### Serviceability Limit State (SLS)

The GL Rules for wind turbine foundation design state that for spread and piled foundations no uplift should exist beneath the foundation base or on the anchors for maximum normal production loading. This implies that the relative eccentricity ratio,  $e/B < 1/6$ , so that base reaction are in compression across the full cap area and  $e/D_a < 1/6$  indicates that no net uplift occurs on the anchors at the anchor circle diameter,  $D_a$ . This is easily satisfied as the foundation is designed

#### Foundation Stiffness Analysis

The stiffness of the rock anchor foundation system is dependent on the combined compressibility of the subgrade beneath the cap and the anchor system and a minor component of the rigidity and compressibility of the cap itself. To compute the stiffness of the anchor system requires the consideration of the stiffness of the various components of the foundation and how they interact with each other. From basic engineering mechanics, the following relationships are developed for springs in series or in parallel to each other.

	Series	Parallel (two springs)
Deflections (and Rotation)	$\delta_T = \delta_1 + \delta_2 + \dots \delta_n$	$\delta_T = \delta_1 = \delta_2$ or $\theta_T = \theta_1 = \theta_2$
Forces (and Moment)	$P_T = P_1 = P_2 = \dots P_n$	$P_T = P_1 + P_2$ or $M_T = M_1 + M_2$
Spring Stiffness	$K_T = 1/(1/K_1 + 1/K_2 + \dots 1/K_n)$	$K_T = K_1/F_1 = K_2/(1-F_1) = K_1 + K_2$ where $F_1 = K_1/(K_1 + K_2)$

And in general,  $K = AE/L$ , and  $\delta = PL/AE$ ,  $\delta = P/K$  where,

$P$  = Force (load),  $M$  = Moment,  $L$  = length of member under stress,  $A$  = Cross sectional area,  $E$  = elastic modulus,  $K$  = spring stiffness,  $\delta$  = deflection,  $\theta$  = rotation, subscript T is total, and subscripts 1 through n are individual elements.

The post-tensioned anchor rod (a) and the Prestress Zone of the subgrade (pz) acts as two springs in parallel to each other when external loads ( $P_T$ ) are applied ( $P_T = P_a + P_{pz}$ ). After post-tensioning the rod, only a portion of the load fluctuations are transmitted to the rod with the remaining load distributed through the bearing area, since the rod stiffness is far less the bearing stiffness (conversely rod elongation is far more than the bearing compression).

So long as the preload is not exceeded, the rod stress remains relatively constant, hence little concern for fatigue and cyclic degradation.

During post-tensioning, several elements are in series to each other, each contributing to the compression deflection as a tension load is elongating the rod. These include:

Bearing Compression of the Top Anchor Plate (minor),  
Local Bearing Compression of the Concrete beneath the Anchor Plate (minor),  
Bearing compression of the Cap base (minor), and in parallel to each other:  
Bearing compression of the Subgrade beneath the Cap (major component)  
Compression of the Grout around the Rock anchor.

Anchor Rod tension causes mobilization of the total anchor perimeter skin friction, elongation of the free stress (unbonded) rod length, and elongation of an effective length along the bonded portion of the anchor. The skin friction and rod elongation are in series to each other.

Local compression (settlement) of the concrete bearing and subgrade compression (settlement) can be computed from the following classical equation:

$$\delta = qBI_p(1-\nu^2)/E$$

Where, B = bearing width, q = contact pressure = P/A,  $\nu$  = Poisson ratio

$I_p$  = vertical strain influence factor that is dependent on the variation of modulus with depth.

$I_p = \pi/4$  and 1 for circular and square disks, respectively on elastic half space

E = elastic modulus, for soils and rock.

Combined Cap Subgrade and Rock anchor System Stiffness: The cap bearing on subgrade and the rock anchors act in parallel to each other with applied overturning moments. So from the preceding relationships, the proportion of moment capacity carried by the anchors can be derived if the anchor and the subgrade are assumed to be independent of each other:

Total moment =  $M_T =$

Moment resisted by anchors ( $M_a$ ) + moment resisted by cap bearing on subgrade ( $M_{pz}$ )

Rotation,  $\theta_{pz} = M_a/K_{\theta a} = M_c/K_{\theta pz}$ ,

Where,  $K_{\theta a}$  and  $K_{\theta pz}$  are the rotational stiffness of the anchor group & prestress zone of cap subgrade, respectively

Contribution of anchor group factor,  $F = M_a/M_T = K_{\theta a}/(K_{\theta a} + K_{\theta c})$

However, there remains additional rotation ( $\theta_{lz}$ ) below the prestressed zone (the lower zone, subscript lz) in series so that the total rotational stiffness of the foundation systems is:

$$K_{\theta T} = [(K_{\theta a} + K_{\theta pz})^{-1} + K_{\theta lz}^{-1}]^{-1}$$

### **Rotational Stiffness of Anchor Group – Simplified Approach**

A general simplified approach to compute rotational stiffness of an anchor group is given Mokwa and Duncan (2003) by:

$$K_{M\theta} = \frac{\Delta M}{\Delta \theta} = \sum_{i=1}^n [K_{\Delta c}(x_i - a)^2] + \sum_{i=1}^n [K_{\Delta t}(x_i - a)^2]$$

But for a circular anchor group, the rotational stiffness of the anchor group is simpler still and may be computed as:

$$K_{M\theta} = nDaKa^2/8$$

Where  $n$  = number of anchors,  $Da$  = diameter of anchor circle, and  $Ka$  = axial stiffness of the anchor. The maximum tension or compression load to the anchor assuming a neutral axis at the centerline is  $P = 4M/nDa$ , where  $M$  = moment carried by rock anchor group. The maximum deflection of the anchor  $\delta a$  may be computed by  $\theta Da/2$

The elastic elongation of the anchor may be computed from:

$$\delta = P/Ka,$$

where  $K_a = Ka$  = axial stiffness of anchor =  $ArEr/La$   
 $Ar$  = cross sectional area of anchor  
 $Er$  = Modulus of Elasticity of Steel  
 $La$  = active length of anchor

The active length of the anchor,  $La$  may be variable, dependent of the nonlinear distribution of transfer of bond stress and skin friction along the length of the anchor. At minimum it is the unbonded length of the anchor,  $Lu$ , but as maximum it may be a portion of the unbonded length plus one half of the mobilized bonded length,  $Lb$ . We have used a 15 foot active length, that is the unbonded length plus one half of the estimated mobilized bonded length with a safety factor of a least two against ultimate bond stress. For prestressed anchors, the axial stiffness of the anchor is the sum of the rod stiffness and subgrade stiffness, so only a portion of the load induced from applied moments is carried by the anchor rod, with the remainder carried by change in compression of the subgrade and bearing plate.

Other minor, but quantifiable, component of deflection is associated with the eccentric loading from the transfer of loads from tower bolt circle to the anchor circle (similar to cantilever action in a very deep beam).

### Stiffness of Cap Subgrade by Elastic Half Space (EHS) Method

Dynamic spring stiffness of the cap may be derived from the classical, closed form equations for an embedded rigid block within an elastic half space (EHS) theory. This method is used commonly for design of machine foundations and spread footings for wind turbines. The dynamic spring stiffness values may be derived using the following published equations from the *Guidelines for Design of Wind Turbines* (DNV and Risø National Laboratory, 2002). These basic equations may be modified by depth to stiffer (rigid) layer of embedment factors subject to the limitations of validity.

Vertical Stiffness at base:  $K_z = 4Gr/(1-\nu)$   
 Horizontal Stiffness at base:  $K_x = 8Gr/(2-\nu)$   
 Rocking Stiffness at base:  $K_\theta = 8Gr^3/[3(1-\nu)]$

***Dynamic Soil or Rock Properties***

The dynamic soil properties are derived for estimated average values of the subsurface profile beneath the foundation. The maximum shear modulus,  $G_o$  is the most relevant property for dynamic analyses. The maximum (low-strain) shear modulus is related to the shear wave velocity,  $V_s$ , and mass density,  $\rho$  by the equation:

$$G_o = \rho V_s^2$$

$$\text{But, also } G = E/[2(1+\nu)]$$

***Stiffness Influence Factors for Layered Media***

By extension of elastic theory we compute the moment/rotation influence factors ( $\Delta I_\theta$  and  $I_\theta$ ) for layered media, analogous to the vertical strain influence factors. In the case of rotations and rotational stiffness we have interchanged elastic modulus, applied pressure and incremental displacements for shear modulus, applied moment and incremental rotations, derived from  $I_\theta$ , respectively. Rotations within each layer are:

$$\theta = M*[8Gr^3/3(1-\nu)]^{-1}*I_\theta$$

Derivation of Moment/Rotation Influence factors ( $\Delta I_\theta$  and  $I_\theta$ ): To ESSW knowledge, there are no published, closed form solutions for the complete layer influence factors at shallow depths for moment/rotation as there are for vertical pressures and strain. However, a reasonable relationship can be developed from what is published and understanding of basic elastic theory.

DNV/Risø (2002) cite the stiff (rigid) layer factor for rotational stiffness as  $N_\theta = 1+R/6H$ , where  $H$  is depth to a stiff (rigid) layer beneath of the foundation, and  $R$  is the radius of the foundation, but with a range of validity of  $H/R > 0.75$ , and cited elsewhere as  $H/R > 2$  (DOD, 1997). Note that this is significantly less than the corresponding stiff (rigid) layer factor for vertical stiffness as  $N_v = 1+1.28R/H$ . This indicates that the majority of the stiffness and conversely rotations occurs nearer to the base of the foundations than vertical displacements.

The cumulative displacement or rotational influence factors ( $I_h$  or  $I_\theta$ ) are merely the inverse of the layer stiffness factors, that is to say,  $I_h = 1/N_v$  and  $I_\theta = 1/N_\theta$ . These layer stiffness and influence factors are plotted in the attached results. Note however, that  $I_\theta = [1+R/6H]^{-1}$  becomes invalid at about  $h/d = 0.9$  as that the maximum  $I_\theta$  factor can not exceed a value of  $[h/d]$ , as this implies no attenuation of moment with depth and is the maximum possible influence factor for shallow depth in the range of invalidity for the equation. We have derive a reasonable equation that smoothly transitions from  $I_\theta = [h/d]$  in the shallow zone to  $[1+R/6H]^{-1}$  in the deep zone. We observe the Boussineq equation for vertical stress, but exchanging exponents will result in a  $\Delta I_\theta$  such that the cumulative  $I_\theta$  relationship will provide such a reasonable and smooth transition.

$$\text{Hence we derive: } \Delta I_\theta = 1 - [1 + (h/r)^4]^{-7}$$

## **Part B - Structural Design and Analysis**

### ***Design Methods***

The structural design analysis for the reinforced concrete pier and the anchor bolts consists of Microsoft Excel spreadsheet calculations that employ traditional design methods of the American Concrete Institute (ACI). The calculations include an input page that records the inputs for the remaining analyses.

The design analysis for the reinforced concrete cap, the anchor bolts, and the rock anchors employed traditional design methods. The concrete design for the anchors was based upon the criterion of the American Concrete Institute (ACI) and the Post-Tensioning Institute (PTI) Recommendations for Prestressed Rock and Soil Anchors.

The rock anchor foundation stiffness and deflections are calculated considering a combination of elastic compression/elongation of the anchors. The stiffness of the foundation system is ultimately calculated using the combined stiffness contributed by the cap on the subgrade based on the elastic half space method, and the calculated stiffness of the anchor group.

### **Design Assumptions**

The calculations make the following assumptions with regard to the behavior of the cap/rock anchor system:

1. Vertical loads are transferred through the base plate to the concrete cap. The cap, in turn, transfers these axial loads to the subgrade and the circumferential rock anchors. The axial loads are then transferred to the soil from the rock anchors via skin friction.
2. Post-tensioning of the Rock anchors creates a compression stress in the subgrade that creates a clamping effect. This clamping effect is used to resist overturning moments in the foundation.
3. Due to the clamping force of the prestressed rock anchors, lateral loads on the cap are resisted by shear friction (or cohesion) between the base of the cap.
4. Overturning moments are resisted by skin friction along the length of the rock anchors.
5. The forces required to post-tension the WTG anchor bolts and the rock anchor bolts are calculated to exceed the load imposed by the maximum wind forces specified by the WTG manufacturer. It is therefore assumed that the normal operational loads of the WTG will not produce stress reversals at either the soil/concrete interface, in the WTG anchor bolts or in the Rock anchors. Further, no cyclic degradation of the soil is assumed to occur.

### **Structural Analysis**

#### ***Cap Analysis and Design***

The basic structural design calculations were performed by P&H. The basic design calculations cover all of the concrete design associated with the cap and the appurtenances required to anchor the wind turbine generator to the cap. The calculations produced by P&H consist of structural calculations that are based on the American Concrete Institute (ACI) design requirements for reinforced concrete. The analyses are all based upon linear elastic theory.

The calculations include analyses of the following items:

- (i) analysis of the cap for flexural and shear capacity and reinforcement,
- (ii) anchor bolt analysis,
- (iii) bearing on grout and concrete, and

**(iv)** shear analysis.

All of the calculations prepared by P&H use the extreme wind load moment as the basis of design. For design based on the extreme wind load, the IEC specifies a partial load factor of 1.1 or 1.35 to be applied to the design load.

The cap analysis is based on the assumption that the cap will be required to resist bending that is induced by the transfer of forces between the tower anchor bolts and the Rock anchors. In this analysis, P&H assumes that the cap acts as a cantilever that is fixed at the location of the tower anchor bolts; the load is assumed to act at the location of the rock anchors. Further, the analysis considers a beam width that is equal to the tributary width of the tower anchor bolts calculated using the average of the anchor bolt and rock anchor bolt circles. Based on the use of the specified reinforcement, the P&H calculation determines the capacity of the concrete section considering the cases of a single layer of reinforcement and also as a doubly reinforced section. For both cases, the ultimate strength of the concrete section is calculated to exceed the applied loading.

The analysis of the tower anchor bolts is accomplished via an analysis that factors the extreme wind loads in generally in accordance with the provisions of the ACI code; for this analysis, the dead load factor is increased above that specified in the reference equation cited by P&H. The force in the anchor bolts is calculated by conservatively combining (absolutely) the tensile load induced by overturning of the tower with the compressive load associated with dead load of the tower and nacelle. The resulting bolt "tension" was compared to the yield strength of the anchor bolts and shown not to exceed said yield strength.

P&H uses the calculations of the bearing stresses on the grout and on the concrete as its basis for the required grout and concrete design strengths. The calculations of the compressive loads under the tower base flange are calculated using the combined compressive loads induced by the dead load and extreme wind load overturning moment. This analysis is based upon the use of factored extreme wind loads; the loads are factored using the same method as discussed in the tower anchor bolt analysis. Additionally, the prescribed post-tension load (unfactored) for the tower anchor bolts was also applied to determine the maximum amount of compression under the tower base flange.

To further confirm the required design strength of the grout beneath the tower base flange and the required concrete design strength, P&H performed an analysis using the serviceability requirements for prestressed concrete of the ACI code. For these calculations, the unfactored service loads (normal operating loads) were used as the basis of the analysis. In this analysis, both the concrete and grout were checked for resistance to forces caused by compression due to prestress plus sustained load and by compression due to prestress plus total load..

The base shear analysis shows that the lateral loads due to the extreme wind can easily be resisted by the anchor bolts themselves, without consideration of the additional capacity provided by the friction beneath the base plate.

The lateral loads are transferred from the cap to the soil and countered by the lateral sliding resistance of the cap due to the effects of dead load (turbine, tower and cap) plus the post-tensioned anchorage loads. In this case, the cap and anchors are clamped together with the soil by the prestressing action. This set of conditions makes it highly unlikely that the soil at the base of the cap would separate from the cap. As a result, P&H assumes that the soil will remain in contact with the bottom face of the foundation. Thus, lateral loads can be transferred to the soil via the mechanism of skin friction/cohesion between the base of the cap and the soil mass below.

**Shear Analysis**

The base shear analysis shows that the lateral loads due to the extreme wind can easily be resisted by the anchor bolts themselves, without consideration of the additional capacity provided by the friction beneath the base plate. The result of this analysis generally indicates that the nominal shear strength of the concrete exceeds applied ultimate shear loading.



### **Anchorage Analysis**

The anchorage analysis uses the bolt forces from the concrete analysis to estimate the compressive forces in the concrete under the base flange. The Anchorage Analysis calculates the total load on the base flange as the force in the bolts (due to the sum of the prestress and the shortening described in the Concrete Analysis), that is, the force that is “clamping” the base flange to the underlying grout and concrete. By subtracting the area of the bolts from the overall area of the flange, the compression stress in the concrete under the base flange is calculated and favorably compared to the stress to the design strength of the concrete.

The Anchorage Analysis also considers the design of the lower end of the anchor bolt where it bears on the embedment ring to create the compression in the concrete. Since the embedment ring has the same area as the base flange, and since the tension in the anchor bolt is the same at the top and at the bottom, the compressive stress in the lower concrete is the same as it is in the top under the base flange. The concrete stress check is equally acceptable.

### **Axial Bearing Analysis**

The calculations include an analysis of the axial concrete strength that includes the appropriate dead overturning and post-tensioning loads. The analysis of the axial loads is based on the requirements of Section 18.4.2 of the ACI Code. In this analysis, the required strength is calculated using the service load level (rather than the extreme wind); additionally, the analysis considers that for the transient loading case, the allowable stress level is adjusted with a one-third increase in the allowed stress.

### **Conclusions**

The Patrick and Henderson rock anchor foundation is a relatively simple and innovative design. As designed, it has adequate capacity against upset from the high overturning moment imposed by the wind turbine structures and provides sufficient rigidity (rocking stiffness). *For turbine sites with rock at depths of 10 feet or less from top of the foundation, 40 foot long rock anchors are adequate and the mud mat should extend to the top of rock. For Turbine Sites T4 and T33 with the depth to rock at about 12.9 and 9.8 feet below expected mud mat elevations, respectively, the anchor lengths should be extended to 50 feet and the leveling course (mud mat need only to be minimum thickness of 6 inches. The required preload for these sites may need to be increased to 435 kips to allow for overconsolidation and subsequent relaxation of the glacial till soil*

### **EARTH SYSTEMS SOUTHWEST**



Shelton L. Stringer, PE\*, GE, PG\*, EG  
Associate Geotechnical Engineer & Engineering Geologist  
Senior Vice President

\* registered in California

Attachments:   References  
                  Appendix A:   Geotechnical Analysis Calculations  
                  Appendix B:   Structural Analysis Calculations

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**APPENDIX A**  
**Geotechnical Analysis Calculations**

**Earth Systems Southwest**

Project: **Stetson Mountain Wind Project** Danforth, ME  
 14 - 40-foot Anchors on 20-foot Ring with 24-foot Diameter Cap  
 using: Williams 2-1/2-in diameter Grade 150 ksi Anchors or Equal  
 Wind Turbine: **GEWE 1.5sle 1.5 MW on 80-m Hub Height Tower**

ESSW Job No.: 09824-55

Date: 09/19/07

**GEOTECHNICAL FOUNDATION ANALYSIS OF P&H ROCK ANCHOR FOUNDATION**

Analysis and Microsoft Excel Spreadsheet Developed by Shelton L. Stringer, PE, GE

**FOUNDATION DIMENSIONS AND PROPERTIES**

Outside Cap (CMP) Diameter	<b>Do</b>	=	<b>24.00 feet</b>	=	7.315 m	
Cap Thickness	<b>Tc</b>	=	<b>5.00 feet</b>	=	1.524 m	
Embedded Cap Depth	<b>Lc</b>	=	<b>4.0 feet</b>	=	1.219 m	
Anchor Diameter	<b>Da</b>	=	<b>5.0 inches</b>	=	0.127 m	
Anchor Pile Length below Cap	<b>La</b>	=	<b>34.5 feet</b>	=	10.516 m	
Anchor Circle Diameter	<b>D</b>	=	<b>20.00 feet</b>	=	6.096 m	
Number of Anchors	<b>n</b>		<b>14</b>			
Concrete Compressive Strength	<b>fc</b>	=	<b>5,000 psi</b>	=	34.5 MPa	
Density of Concrete	<b>γc</b>	=	<b>150 pcf</b>	=	23.6 kN/m <sup>3</sup>	<u>Equations:</u>
Concrete Modulus of Elasticity	<b>Ec</b>	=	<b>4,030,000 psi</b>	=	27,800 MPa	$57,000 * f_c^{0.5}$
Base Area of Footing	<b>Af</b>	=	<b>452.39 sf</b>	=	42.03 m <sup>2</sup>	$\pi D^2 / 4$
Weight Concrete	<b>Wc</b>	=	<b>339.3 kips</b>	=	1,509 kN	$Af * L * \gamma_c$
Volume Concrete	<b>Vc</b>	=	<b>83.8 cy</b>	=	57.31 m <sup>3</sup>	$Af * L / 27 (cy)$

**GEOTECHNICAL DATA**

	Description	Estimated GSI	Unit Weight		Friction Angle $\phi$ (deg)	Mass Strength		Compressive Strength	
			$\gamma_T$ (pcf)	kN/m <sup>3</sup>		$q_m$ (ksf)	(MPa)	$qu$ (psi)	(MPa)
	Till		130	20	40				
	Weathered Bedock	35	150	24		25	1.2	1450	10
GWT:	GWT - Design Groundwater Table								

**FOUNDATION LOADS**

Extreme Loading	Based on: Foundation Data for GW 1.5sle, IEC IIA 80 m Hub Height, DLC 6.1j					
Horizontal Load	<b>Fxy</b>	=	<b>117.665 kips</b>	=	<b>523 kN</b>	exclusive of partial safety factors
Axial Dead Load	<b>Fz</b>	=	<b>439.637 kips</b>	=	<b>1,956 kN</b>	
Bending Moment at Base	<b>Mxy</b>	=	<b>25,746.608 ft-kips</b>	=	<b>34,908 kN-m</b>	
Height of Tower	<b>h</b>	=	<b>262.47 feet</b>	=	<b>80.00 m</b>	
Effective Application Height	<b>ez</b>	=	<b>218.81 feet</b>	=	<b>66.69 m</b>	

**CALCULATION OF ROCK ANCHOR PULLOUT CAPACITY**

Project: Stetson Mountain Wind Project  
 14 - 40-foot Anchors on 20-foot Ring with 24-foot Diameter Cap  
 using: Williams 2-1/2-in diameter Grade 150 ksi Anchors or Equal

Calculated by: Shelton L, Stringer, PE, GE Date: 9/19/2007

<b>GIVEN:</b>		Foundation Data for GW 1.5sle, IEC IIA 80 m Hub Height, DLC 6.1j		
<b>Horizontal Load</b>	<b>F<sub>x</sub></b>	<b>118 kips</b>	<b>523.4 kN</b>	
<b>Vertical Load</b>	<b>F<sub>z</sub></b>	<b>440 kips</b>	<b>1955.6 kN</b>	
<b>Design Overturning Moment</b>	<b>M<sub>D</sub></b>	<b>25747 ft-kips</b>	<b>34907.7 kN-m</b>	
<b>Check Bonded Length Pullout Capacity</b>		Reference: PTI (1996)		Equations
Anchor Body Diameter	db	5.0 in	0.42 ft	
<b>Bonded Length in Rock</b>	<b>Lb</b>	<b>28.5 feet</b>		
<b>Ultimate Bond Strength</b>	<b>T<sub>w</sub></b>	<b>150 psi</b>		Fractured, Volcanic or Pelite rock
<b>Anchor Pullout Capacity</b>	<b>P<sub>a</sub></b>	<b>806 kips</b>		
<b>Design Load per anchor</b>	<b>F<sub>p</sub></b>	<b>319 kips</b>		
<b>Factor of Safety against Pullout</b>	<b>FS</b>	<b>2.5</b>		
<b>Check Free Stress Stress Length &amp; Pullout Capacity of rock mass (individual &amp; group)</b>				
<b>Input</b>		References: FHWA-IF-99-915 & USCOE EM 1110-1-2906		
Anchor Tilt	α	90 degrees	from horizontal	
<b>Total Unbonded Length</b>	<b>L<sub>u</sub></b>	<b>10.0 feet</b>		
Anchor Length above cap	L <sub>a</sub>	1.5 feet		
<b>Total Length of Anchor</b>	<b>L<sub>t</sub></b>	<b>40.0 feet</b>		
Anchor Circle Dia	D <sub>a</sub>	20.00 ft	ra	Lu + Lb + La 10.21 feet = Da/2 + db/2
No. of Anchors	n	14		
Spacing between anchors	x	4.49 ft		πDa/n
Potential Failure Starts at		0.5 Lb, middle of bonded length for individual anchor		
<b>Soil/Rock Info:</b>		Till	over	Rock
Half Cone Shear Angle	φ	40 deg	60 deg	
Cohesion/Shear Strength	c	0 psf	0 psf =	rock mass cohesion intercept
Moist Unit Weight	γ	130 pcf	170 pcf	
Depth to Top of Rock or free stress	z <sub>R</sub>	10.0 feet		
Groundwater Depth	GWT	100.0 feet		
		Depth	Unit Wt	Pressure
Vertical Soil Pressure	z (feet)	γ (pcf)	σ' <sub>v</sub> (ksf)	q <sub>rm</sub> = factored rock mass compressive strength
at Cap Base	4.0	130.0	0.52	
at bottom of 1st layer or GWT	10.0	130.0	1.30	
at top of rock, z <sub>R</sub>	10.0	130.0	1.30	
Effective Pressure @ top of rock	σ' <sub>v</sub>		1.30	9.03 psi
<b>Calculations</b>		Individual	Group	
Effective Length	L <sub>eff</sub>	24.3 feet	38.5 feet	Lu + (0.5Lb- individual, Lb - group)
Total Overburden Depth	Z <sub>t</sub>	24.3 feet	38.5 feet	L <sub>eff</sub> *sin(α)
Eff. Embedment Depth into Rock	Z	14.3 feet	28.5 feet	Z <sub>t</sub> - z <sub>R</sub>
Radius of Cone of Rock	R	24.7 feet	49.4 feet	Z*tanφ
Area @ top of Cone (individual)	A <sub>c</sub>	1913.8 sf		πR <sup>2</sup>
Area @ top of Cone (group)	A <sub>cg</sub>		796.3 sf/anchor	π(R+ra) <sup>2</sup> /n
Volume of Rock Cone (individual)	V <sub>c</sub>	9090.6 cf		1/3πR <sup>2</sup> Z
Volume of Rock Cone group)	V <sub>cg</sub>		6491.1 cf/anchor	1/3πZ(R <sup>2</sup> +ra <sup>2</sup> +raR)/n
Weight of Rock Cone (individual)	W <sub>c</sub>	1545.4 kips		V <sub>c</sub> *γ <sub>R</sub>
Weight of Cone (group)	W <sub>cg</sub>		1103.5 kips/anchor	V <sub>cg</sub> *γ <sub>R</sub>
Surface Area of Cone (individual)	A <sub>s</sub>	2209.9 sf		πRZ/cosφ
Surface Area of Cone (group)	A <sub>sg</sub>		762.0 sf/anchor	π(R+ra)(Z/cosφ)/n
Shearing Resistance (individual)	S	0.0 kips		A <sub>s</sub> *f <sub>a</sub>
Shearing Resistance (group)	S <sub>g</sub>		0.0 kips/anchor	A <sub>sg</sub> *f <sub>a</sub>
<b>Pullout Capacity (individual)</b>	<b>P</b>	<b>4033 kips</b>		W <sub>c</sub> + S + σ' <sub>v</sub> *A <sub>c</sub>
<b>Pullout Capacity (group)</b>	<b>P<sub>g</sub></b>	<b>2139 kips/anchor</b>		W <sub>cg</sub> + S <sub>g</sub> + σ' <sub>v</sub> *A <sub>cg</sub>
<b>Preload Load per anchor</b>	<b>F<sub>p</sub></b>	<b>319 kips</b>		
<b>Factor of Safety against Pullout</b>	<b>FS</b>	<b>12.64 individual</b>		
<b>Factor of Safety against Pullout</b>	<b>FS</b>	<b>6.70 group</b>		

**CALCULATION OF ROCK ANCHOR PULLOUT CAPACITY**

Project: Stetson Mountain Wind Project  
 14 - 40-foot Anchors on 20-foot Ring with 24-foot Diameter Cap  
 using: Williams 2-1/2-in diameter Grade 150 ksi Anchors or Equal

Calculated by: Shelton L, Stringer, PE, GE Date: 9/19/2007

<b>GIVEN:</b>		Foundation Data for GW 1.5sle, IEC IIA 80 m Hub Height, DLC 6.1j		
<b>Horizontal Load</b>	<b>F<sub>x</sub></b>	<b>118 kips</b>	<b>523.4 kN</b>	
<b>Vertical Load</b>	<b>F<sub>z</sub></b>	<b>440 kips</b>	<b>1955.6 kN</b>	
<b>Design Overturning Moment</b>	<b>M<sub>D</sub></b>	<b>25747 ft-kips</b>	<b>34907.7 kN-m</b>	
<b>Check Bonded Length Pullout Capacity</b>		Reference: PTI (1996)		Equations
Anchor Body Diameter	db	5.0 in	0.42 ft	
<b>Bonded Length in Rock</b>	<b>Lb</b>	<b>30.1 feet</b>		
<b>Ultimate Bond Strength</b>	<b>T<sub>w</sub></b>	<b>150 psi</b>		Fractured, Volcanic or Pelite rock
<b>Anchor Pullout Capacity</b>	<b>P<sub>a</sub></b>	<b>852 kips</b>		
<b>Design Load per anchor</b>	<b>F<sub>p</sub></b>	<b>435 kips</b>		
<b>Factor of Safety against Pullout</b>	<b>FS</b>	<b>2.0</b>		For T-4 & T-33
<b>Check Free Stress Stress Length &amp; Pullout Capacity of rock mass (individual &amp; group)</b>				
<b>Input</b>		References: FHWA-IF-99-915 & USCOE EM 1110-1-2906		
Anchor Tilt	α	90 degrees	from horizontal	
<b>Total Unbonded Length</b>	<b>L<sub>u</sub></b>	<b>18.4 feet</b>	Maximum at T-4	
Anchor Length above cap	L <sub>a</sub>	1.5 feet		
<b>Total Length of Anchor</b>	<b>L<sub>t</sub></b>	<b>50.0 feet</b>		
Anchor Circle Dia	D <sub>a</sub>	20.00 ft	ra	Lu + Lb + La 10.21 feet = Da/2 + db/2
No. of Anchors	n	14		
Spacing between anchors	x	4.49 ft	πDa/n	
Potential Failure Starts at		0.5 Lb, middle of bonded length for individual anchor		
<b>Soil/Rock Info:</b>		Till	over	Rock
Half Cone Shear Angle	φ	40 deg	60 deg	
Cohesion/Shear Strength	c	0 psf	0 psf =	rock mass cohesion intercept
Moist Unit Weight	γ	130 pcf	170 pcf	
Depth to Top of Rock or free stress	z <sub>R</sub>	18.4 feet		
Groundwater Depth	GWT	100.0 feet		
		Depth	Unit Wt	Pressure
Vertical Soil Pressure	z (feet)	γ (pcf)	σ' <sub>v</sub> (ksf)	q <sub>rm</sub> = factored rock mass compressive strength
at Cap Base	4.0	130.0	0.52	
at bottom of 1st layer or GWT	18.4	130.0	2.39	
at top of rock, z <sub>R</sub>	18.4	130.0	2.39	
Effective Pressure @ top of rock	σ' <sub>v</sub>		2.39	16.58 psi
<b>Calculations</b>		Individual	Group	
Effective Length	L <sub>eff</sub>	33.4 feet	48.5 feet	Lu + (0.5Lb- individual, Lb - group)
Total Overburden Depth	Z <sub>t</sub>	33.4 feet	48.5 feet	L <sub>eff</sub> *sin(α)
Eff. Embedment Depth into Rock	Z	15.1 feet	30.1 feet	Z <sub>t</sub> - z <sub>R</sub>
Radius of Cone of Rock	R	26.1 feet	52.2 feet	Z*tanφ
Area @ top of Cone (individual)	A <sub>c</sub>	2140.4 sf		πR <sup>2</sup>
Area @ top of Cone (group)	A <sub>cg</sub>		874.1 sf/anchor	π(R+ra) <sup>2</sup> /n
Volume of Rock Cone (individual)	V <sub>c</sub>	10752.0 cf		1/3πR <sup>2</sup> Z
Volume of Rock Cone group)	V <sub>cg</sub>		7580.4 cf/anchor	1/3πZ(R <sup>2</sup> +ra <sup>2</sup> +raR)/n
Weight of Rock Cone (individual)	W <sub>c</sub>	1827.8 kips		V <sub>c</sub> *γ <sub>R</sub>
Weight of Cone (group)	W <sub>cg</sub>		1288.7 kips/anchor	V <sub>cg</sub> *γ <sub>R</sub>
Surface Area of Cone (individual)	A <sub>s</sub>	2471.5 sf		πRZ/cosφ
Surface Area of Cone (group)	A <sub>sg</sub>		844.2 sf/anchor	π(R+ra)(Z/cosφ)/n
Shearing Resistance (individual)	S	0.0 kips		A <sub>s</sub> *fa
Shearing Resistance (group)	S <sub>g</sub>		0.0 kips/anchor	A <sub>sg</sub> *fa
<b>Pullout Capacity (individual)</b>	<b>P</b>	<b>6937 kips</b>		W <sub>c</sub> + S + σ' <sub>v</sub> *A <sub>c</sub>
<b>Pullout Capacity (group)</b>	<b>P<sub>g</sub></b>	<b>3375 kips/anchor</b>		W <sub>cg</sub> + S <sub>g</sub> + σ' <sub>v</sub> *A <sub>cg</sub>
<b>Preload Load per anchor</b>	<b>F<sub>p</sub></b>	<b>435 kips</b>		
<b>Factor of Safety against Pullout</b>	<b>FS</b>	<b>15.95 individual</b>		
<b>Factor of Safety against Pullout</b>	<b>FS</b>	<b>7.76 group</b>		



**PATRICK & HENDERSON TENSIONLESS PIER FOUNDATION**

Project: **Stetson Mountain Wind Project**

Wind Turbine: **GEWE 1.5sle 1.5 MW on 80-m Hub Height Tower**

ESSW Job No.: 09824-55

Date: 09/19/07

**DYNAMIC FOUNDATION ANALYSIS** **by Shelton L. Stringer, PE, GE**

Reference: DNV/Risø (2002) Guidelines for Design of Wind Turbines

<b>Foundation Dimensions &amp; Weights</b>			
Outer Radius	$r_o =$	12.0 feet	= 144 in. = 3.66 m
<b>Rock Properties</b>		<b>Bearing on Rock</b>	
Density	$\gamma =$	170 pcf	= 26.7 kN/m <sup>3</sup>
Mass Density	$\rho = \gamma/g$	= 0.000255 lb-sec <sup>2</sup> /in <sup>4</sup>	= 2,720 kg/m <sup>3</sup>
Intact Compressive Strength	$qu =$	6200 psi	= 42.7 MPa
Geological Strength Index	$GSI =$	35	$Em = (qu/100)^{0.5} * 10^{[(GSI-10)/40]}$ (GPa) D = 0
Young's Rock Mass Modulus	$Em =$	= 406,000 psi	= 2,800 MPa
Avg. Shear Modulus	$G = Es/2(1+\nu)$	= 156,000 psi	= 1,080 MPa
Avg. Shear Wave Velocity	$V_{so} = (G/\rho)^{0.5}$	= 2060 ft/sec	= 628 m/sec
Strain Reduction Ratio	$G/Go =$	0.80	for rock subgrade or very dense soil
Shear Modulus	$G = Go * G/Go$ Ratio	= 124,800 psi	= 860 MPa
Poisson Ratio	$\nu =$	0.30	
<u>Dynamic Spring Constants</u>			
Vertical	$Kz = 4G * r_o(1-\nu)$	= 102,693 kip/in	= 18.0 GN/m
Horizontal	$Kx = 8G * r_o/(2-\nu)$	= 84,570 kip/in	= 14.8 GN/m
<b>Rocking</b>	$K\theta = 8G * r_o^3/[3(1-\nu)]$	= 1.42E+12 lb-in/rad.	= <b>160 GN-m/rad.</b>

**CALCULATION OF ROTATIONAL STIFFNESS CONSIDERING PRESTRESS EFFECTS OF ANCHORS**

Project: **Stetson Mountain Wind Project**

Reference: DNV/Risø (2002) Guidelines for Design of Wind Turbines

Foundation and Layer Dimesions		Assume T4 conditions -13 feet till beneath foundation				
Diameter of Foundation	d	24.00	feet		7.315	m
Thickness of Foundation	t	5.00	feet		1.524	m
Base Area	A	452.4	sf			
Radius of Foundation	a (r)	12.00	feet		3.658	m
Weight of Foundation	WT	339	kips		1509	kN
Embedment Depth to Subgrade	z <sub>E</sub>	5.00	feet		1.524	m
Prestress Depth Zone	z <sub>PZ</sub>	20.00	feet		6.096	m
Stiff Layer Depth	h	50.0	feet			h/r: 3.75
<b>Anchor Properties &amp; Preload</b>						
Anchor Circle Diameter	Da	20	feet		6.096	m
Deisign Prestress per Anchor	Pta	368	kips		1637	kN
Number of Anchors	n	14				
Prestress Load to Foundation	Ppt	5152	kips		22917	kN
External Applied Load	Pext	440	kips		1958.9	kN
Total Load to Base of Foundation	Pt	5932	kips		26385	kN
Anchor Hole Diameter	d	5.0	inches		127	mm
Anchor Rod Diameter	da	2.57	inches		65	mm
Area of Anchor	Ar	5.19	in <sup>2</sup>		3347	mm <sup>2</sup>
Modulus of Anchor	Er	29000000	psi		200	GPa
Active Length of Anchor	La	20.00	feet		6.096	m
Moment Resistance Developed by Anchors:	Mr	25760	ft-kips		34926	kN-m
Anchor Axial Stiffness	K <sub>a</sub>	627	kip/in			nD*Pta/4 ArEr/La
<b>Prestressed Zone (PZ) Stresses</b>						
Base Pressure	q (Δσ <sub>v</sub> )	13.11	ksf		91	psi
Unit Weight in Prestressed Zone	γ	130	pcf			
Insitu Vertical Stress at Prestress Zone	σ <sub>v</sub>	11.3	psi		78	kPa
Operational Modulus Factor in PZ	lo	3.0			Input or [(σ <sub>v</sub> + Δσ <sub>v</sub> )/σ <sub>v</sub> ] <sup>0.5</sup>	
<b>Layer Properties</b>		G/Gmax	0.80		G' =G*G/Gmax*lo	
I <sub>θ</sub> (pz) = (z <sub>PZ</sub> - z <sub>E</sub> )/d	Influence	Stiffness	Elastic	Poisson's	Shear	Operational
I <sub>θ</sub> (lz) = I <sub>θ</sub> (h) - I <sub>θ</sub> (pz)	Factor	Factor	Modulus	Ratio	Modulus	Shear Modulus
I <sub>θ</sub> (h) = 1/(DNV stiff layer factor)	I <sub>θ</sub>	N <sub>θ</sub> = 1/I <sub>θ</sub>	E (ksi)	ν	G (ksi)	G' (psi)
Prestress Zone (pz)	<b>0.900</b>	1.111	50	0.30	19	46
Below Prestress Zone (lz)	<b>0.091</b>	10.932	400	0.30	154	123
at Stiiff Layer (h):	<b>0.991</b>	1.009	400	0.30	154	154
<b>Anchor Group Rotational Stiffness</b>	<b>K<sub>θa</sub></b>	5267850	kip-ft/rad		nDa <sup>2</sup> Ka/8	
		7.1	GN-m/rad			
Prestressed Subgrade Rotational Stiffness	K <sub>θps</sub>	66.1	GN-m/rad		8/3G' <sub>pz</sub> r <sup>3</sup> /(1-ν)*Nθ	
Prestressed Zone Rotational Stiffness	K <sub>θpz</sub>	73.3	GN-m/rad		K <sub>θa</sub> + K <sub>θpz</sub>	
Lower Subgrade Rotational Stiffness	K <sub>θlz</sub>	1728.1	GN-m/rad		8/3G' <sub>lz</sub> r <sup>3</sup> /(1-ν)*Nθ	
<b>Total Rotational Stiffness</b>	<b>K<sub>θT</sub></b>	<b>70.3</b>	<b>GN-m/rad</b>		<b>[K<sub>θpz</sub><sup>-1</sup>+K<sub>θlz</sub><sup>-1</sup>]<sup>-1</sup></b>	
<b>Operational Moment</b>	<b>Mxy</b>	<b>12978</b>	<b>ft-kips</b>		<b>17596 KN-m</b>	
Total Rotation	θ	0.00025	radians		Mxy/Kθ <sub>T</sub>	
Average Shear Strain	γ	0.013%			θ/2	
Relative Rotation of Anchor Group	θ	0.00024	radians		Mxy/Kθ <sub>pz</sub>	
Elongation of Anchor	δa	0.029	inches		0.73	mm
Change in Rod Force	ΔFa	18	kips		80	kN
<b>Factor of Increase in Anchor</b>		<b>1.05</b>			<b>[Pta+ΔFa]/Pta</b>	





Earth Systems Southwest

**OVERTURNING STABILITY ANALYSIS OF P&H ANCHOR FOUNDATION**

Project: **Stetson Mountain Wind Project**

14 - 40-foot Anchors on 20-foot Ring with 24-foot Diameter Cap

using: Williams 2-1/2-in diameter Grade 150 ksi Anchors or Equal

Turbine: GEWE 1.5sle 1.5 MW on 80-m Hub Height Tower

Anchor: Williams		2-1/2	
fy	120 ksi		
fu	150 ksi		
d	2.50 inch	64 mm	
Ar	5.19 in <sup>2</sup>	3348 mm <sup>2</sup>	
<b>Fu</b>	<b>778 kips</b>	<b>2769 kN</b>	
Fy	622 kips	3461 kN	
0.6Fu	467 kips		
P-allow.:	<b>467 kips</b>		

Outer Anchor Diameter	Dc-o	5.00 inches	127 mm
Outer Rod Diameter	Dro	2.56 inches	65 mm

Concrete/Grout Strength	f <sub>c</sub>	3000 psi	20.7 MPa
Concrete/Grout Modulus	E <sub>c</sub>	3120000 psi	21,500 MPa
Steel Modulus	E <sub>s</sub>	29000000 psi	199,900 MPa
Rod Area	A <sub>r</sub>	5.19 sq-in	3348 mm <sup>2</sup>
Anchor Circle Diameter	D <sub>a</sub>	20.00 feet	6.096 m
Number of Anchors	n	14	

<b>Min. PreLoad per Anchor</b>	<b>Pta</b>	<b>319 kips</b>	<b>1419 kN</b>
Prestress Load to Foundation	Ppt	4466 kips	19866 kN

<b>Vertical Load</b>	<b>Fz</b>	<b>440.4 kips</b>	<b>1958.9 kN</b>
Foundation Weight	W	339.3 kips	1509.2 kN

Total Load to Base of Foundation	Pt	5246 kips	23334 kN	Ppt+Fz+W
Active Length of Anchor	La	15.00 feet	4.572 m	
Resistance Developed by Anchors:	Mr	22330 ft-kips	30275 kN-m	n*Pta/4
Anchor Axial Stiffness	K <sub>a</sub>	836 kip/in	146435 kN/m	ArEr/La
Preload Elongation of Anchor	δ <sub>p</sub>	0.382 inches	9.69 mm	Pta/K <sub>a</sub>
Anchor Group Rotational Stiffness	K <sub>0a</sub>	7.02E+06 kip-ft/rad	9.5 GN-m/rad	nDa <sup>2</sup> K <sub>a</sub> /8
Subgrade Rotational Stiffness in PZ	K <sub>0ps</sub>	5.16E+07 kip-ft/rad	70.0 GN-m/rad	8/3G' <sub>pz</sub> <sup>3</sup> /(1-ν) <sup>2</sup> N <sub>0</sub>
Prestressed Zone Rotational Stiffness	K <sub>0pz</sub>	5.87E+07 kip-ft/rad	79.5 GN-m/rad	K <sub>0a</sub> + K <sub>0pz</sub>
Diameter (width) of Foundation	B	24.00 feet	7.315 m	
Thickness of Foundation	T	5.00 feet	1.524 m	

Load Case:	4.2	Lift-off				2
Blade	DLC	LF	Fz	Fr	Mr	% Mext
LM37.3p2	1	1.00	-1958.9	219	17596.2	50%

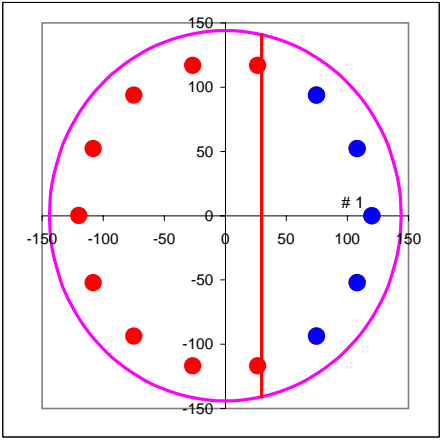
Sliding Stability Check					
Friction	Coeff.	Ff	FS	FS = Ff/Fxy	
φ	μ	kips		Ff = μ*Pt	
40	0.76	3961.5	80.5	Okay	μ = 0.9tan φ

7.38E+05 kip-ft/rad to 1 GNm/rad  
from Rotational Stiffness Worksheet

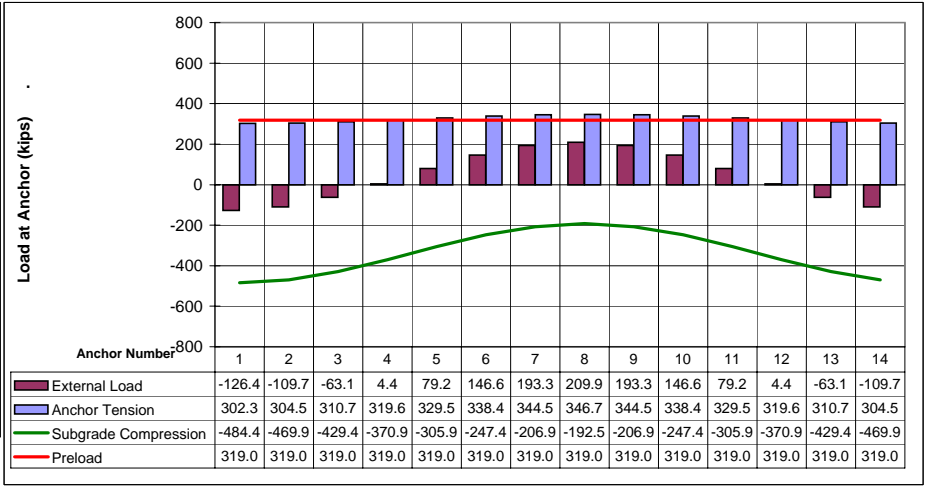
<b>Moment</b>	<b>Mxy</b>	<b>12978.3 ft-kips</b>	<b>17596.2 KN-m</b>	50% M-extreme
<b>Horizontal Load</b>	<b>Fxy</b>	<b>49.2 kips</b>	<b>219.0 KN</b>	
Moment at Base	Mb	13224.5 ft-kips	17930.0 KN-m	Mxy + Fxy(T)
Rotation at Base	θ	0.00022 radians		Mb/Kθ <sub>pz</sub>
Max Force to Anchor (e=0)	Fa	188.9 kips	840.4 kN	4M/nD

Moment Capacity and Stability of Anchor System						
Eccentricity	e	2.48 feet	0.757 m	Mb/Vt	Vt = Σ(T+DL)	Relative Eccentricity e/B
<b>Stability Ratio (Factor of Safety): SR (FS)</b>	<b>4.83</b>			e <sub>max</sub> /e	e <sub>max</sub> = 0.5B	<b>0.104 &lt; 0.167 Okay</b>

No of Rows: 1		θDa/2		δa*Ka		P		T		C		(Fz+W)/n		[Pta+ΔFa]/Pta		DL*x	
No. of Anchors	Offset Distance (feet)	Offset Position	x Distance (in)	y Distance (in)	x Distance (feet)	δa Anchor Deflection (inches)	ΔFa Change in Rod Force (kips)	External Load @ Anchor (kips)	Tension in Rod (kips)	Tributary Subgrade Compression (kips)	DL Dead Load (kips)	Pt Preload (kips)	Factor of Increase or Decrease	P*x External Moment (ft-kips)	T*x Anchor Moment (ft-kips)	Dead Load Moment (ft-kips)	
1	0	0	120.00	0.00	7.52	-0.0200	-16.7	-126.4	302.3	-484.4	55.7	319.0	0.948	-950	-2272	-418.5	
1	0	0	108.12	52.07	6.53	-0.0173	-14.5	-109.7	304.5	-469.9	55.7	319.0	0.955	-716	-1987	-363.4	
1	0	0	74.82	93.82	3.75	-0.0100	-8.3	-63.1	310.7	-429.4	55.7	319.0	0.974	-237	-1165	-208.9	
1	0	0	26.70	116.99	-0.26	0.0007	0.6	4.4	319.6	-370.9	55.7	319.0	1.002	-1	83	14.4	
1	0	0	-26.70	116.99	-4.71	0.0125	10.5	79.2	329.5	-305.9	55.7	319.0	1.033	-373	1552	262.3	
1	0	0	-74.82	93.82	-8.72	0.0232	19.4	146.6	338.4	-247.4	55.7	319.0	1.061	-1278	2950	485.6	
1	0	0	-108.12	52.07	-11.49	0.0305	25.5	193.3	344.5	-206.9	55.7	319.0	1.080	-2222	3960	640.1	
1	0	0	-120.00	0.00	-12.48	0.0331	27.7	209.9	346.7	-192.5	55.7	319.0	1.087	-2621	4329	695.3	
1	0	0	-108.12	-52.07	-11.49	0.0305	25.5	193.3	344.5	-206.9	55.7	319.0	1.080	-2222	3960	640.1	
1	0	0	-74.82	-93.82	-8.72	0.0232	19.4	146.6	338.4	-247.4	55.7	319.0	1.061	-1278	2950	485.6	
1	0	0	-26.70	-116.99	-4.71	0.0125	10.5	79.2	329.5	-305.9	55.7	319.0	1.033	-373	1552	262.3	
1	0	0	26.70	-116.99	-0.26	0.0007	0.6	4.4	319.6	-370.9	55.7	319.0	1.002	-1	83	14.4	
1	0	0	74.82	-93.82	3.75	-0.0100	-8.3	-63.1	310.7	-429.4	55.7	319.0	0.974	-237	-1165	-208.9	
1	0	0	108.12	-52.07	6.53	-0.0173	-14.5	-109.7	304.5	-469.9	55.7	319.0	0.955	-716	-1987	-363.4	
14			Sum:				77	585	4543	-4738	780	4466	14.242	-13224	12842	1937	



Anchors - Red Applied Tension, Blue- Compression









**APPENDIX B**  
**Structural Analysis Calculations**

**PATRICK & HENDERSON PILE ANCHOR FOUNDATION**

Project: **Stetson Mountain Wind Project**  
 Wind Turbine: **GEWE 1.5sle 1.5 MW on 80-m Hub Height Tower**  
 Location: **Danforth, ME**  
 ESSW Job No.: **09824-55** P&H Job No.: **07-036**  
 Date: **09/19/07**

**FOUNDATION DIMENSIONS**

<b>Cap Outside Diameter</b>	<b>Do</b>	=	<b>24.0</b>	feet	=	7.315 m
<b>Foundation Cap Thickness</b>	<b>T</b>	=	<b>5.0</b>	feet	=	1.524 m
Concrete Compressive Strength	<b>f<sub>c</sub></b>	=	<b>6,000</b>	psi	=	41 MPa
Volume of Concrete Cap	<b>V<sub>cap</sub></b>	=	84	cy	=	64 m <sup>3</sup>

**Pile Anchors**

<b>Number of Anchors</b>	<b>n</b>	=	<b>14</b>				
Outer Anchor Circle Diameter	<b>Do</b>	=	<b>20.0</b>	feet	=	6.096 m	
Inner Anchor Circle Diameter	<b>Di</b>	=	<b>0.0</b>	feet	=	0.000 m	
Pile Anchor Diameter	<b>Da</b>	=	<b>24</b>	inches	=	0.6 m	
<b>Anchor Post Tension Force</b>	<b>Pt</b>	=	<b>435</b>	kips	=	1935 kN	
Anchor Bolt - Strength	<b>Fu</b>	=	<b>100,000</b>	psi	=	689 MPa	
Anchor Bolt - Length	<b>La</b>	=	<b>45.0</b>	feet	=	13.7 m	630 LF total
Anchor Bolt - Nominal Size	<b>d</b>	=	<b>2.5</b>	inches	=	Williams Grade 150, 2.5"	
Anchor - Grout Strength	<b>f<sub>cg</sub></b>	=	<b>3,000</b>	psi	=	21 MPa	

**Tower Anchor Bolts**

<b>Specified Post Tension of Bolts</b>	<b>Ps</b>	=	<b>76,200</b>	lbs	=	525 MPa
Number of Tower Bolts	<b>nb</b>	=	<b>140</b>		=	Williams Grade 80, 1-1/4"
Nominal Diameter of Tower Bolts	<b>db</b>	=	<b>1.38</b>	inches	=	35 mm
Length of Bolts	<b>Lb</b>	=	<b>67</b>	inches	=	1.702 m

<b>Reinforcement Schedule</b>	Bar Size	No. LF/ea	No. Pieces	LF	Weight/ea (lbs)	Weight (lbs)	Weight (kN)
<b>Radial Steel</b>							
Type A	10	16	35	560	69	2410	10.72
Type B	10	16	35	560	69	2410	10.72
Type C	10	15	35	525	65	2259	10.05
Type D	10	15	35	525	65	2259	10.05
Stirrups	10	7.50	70	525	32	2259	10.05
Circumferential Hoops	9	57.2	18	1030		3501	15.57
<b>Center Mat</b>							
Top	10	10.1	36	363.6		1565	6.96
Bottom	10	10.1	36	363.6		1565	6.96
<b>Total</b>				<b>4452</b>		<b>18226</b>	<b>81.1</b>

**FOUNDATION LOADS** Based on: Preliminary Loading Data for MWT95/2.4-80m with IEC Class IIA

At base of Tower				Axial	Horizontal	Moment						
Load Case	Description	DLC	Load Factor	Fz KN	Fxy KN	Mxy KN-m	Fz kips	Fxy kips	Mxy kip-ft	% Mext	Load Case	Blade
1	Extreme Load	6.1j	1.00	-1845	523	34,908	-414.7	117.7	25,747	100%	1	GE37c
2	Lift-off	1.0	1.00	-1959	219	17,596	-440.4	49.2	12,978	50%	2	LM37.3p2
3	Tilting	6.1j	1.00	-1845	523	34,908	-414.7	117.5	25,747	100%	3	GE37c
4	Sliding	6.1j	1.00	-1843	523	34,863	-414.4	117.7	25,714	100%	4	GE37c
5	Shear Failure (Beari	6.1j	1.35	-2490	706	47,125	-559.8	158.6	34,758	135%	5	GE37c
6	Tension Loading	1.1	1.00	-1955	327	24,238	-439.6	73.6	17,877	69%	6	LM37.3p2
7												



Patrick & Henderson Anchor Foundation Stetson Mountain Wind Project		GEWE 1.5sle 1.5 MW on 80-m Hub Height Tower				
Code References		based on 2006 IBC & ACI 318				
1.0 Design Input		Equations or Notes	Symbol	Imperial Units	Metric Units	
Item	Cap			Value	Unit	Value Unit
1	Cap Inside Diameter		Di	0.00	feet	0 mm
2	Cap Outside Diameter		Do	24.00	feet	7315 mm
3	Thickness of Foundation Cap		T	5.00	feet	1524 mm
4	Inside Bolt Diameter		Bi	13.5499	feet	4130 mm
5	Outside Bolt Diameter		Bo	14.5210	feet	4426 mm
6	Inside Base Flange Diameter		Id	13.1234	feet	4000 mm
7	Outside Base Flange Diameter		Od	14.9471	feet	4556 mm
8	Diameter of Tower Wall		Dwall	14.0715	feet	4289 mm
9	Width of Base Flange on Tower		bf	10.942	inch	278 mm
10	Inside Width		bfi	4.350	inch	111 mm
11	Base Flange Thickness		Bft	2.95	inch	75 mm
12	Embedment Plate Width		Pw	10.942	inch	278 mm
13	Diameter of Base Flange Bolt Hole		d	1.500	inch	39 mm
14	Nominal Diameter of Bolts	Williams Grade 80, 1-1/4"	db	1.375	inch	35 mm
15	Area of Bolts		ab	1.27	sq-in	819
16	Number of Tower Bolts		nb	140		140
17	Yield Stress of Bolts		fy	75,000	psi	517 MPa
18	Ultimate Stress of Bolts	Grade 75	fpu	100,000	psi	689 MPa
19	Ultimate Strength of Bolts		Fu	127,000	lbs	565 KN
20	Specified Post Tension of Bolts		Ps	76,200	lbs	525 MPa
21	Yield Strength Tower Base Plate	Grade 50	fyb	50,000	psi	345 MPa
22	Rebar Yield Strength	Grade 60	fy	60,000	psi	414 MPa
23	Grout beneath Flange - Strength		f'g	8,500	psi	59 MPa
24	Cap Concrete Compressive Strength		f'c	6,000	psi	41 MPa
25	Concrete Modulus of Elasticity	$E_c = 57,000 * f'c^{0.5}$	Ec	4,415,000	psi	30,400 MPa
26	Steel Modulus of Elasticity		Es	29,000,000	psi	200 GPa
27	Density of Concrete		$\gamma_c$	150	pcf	
28	CMP (3 x 1 Helical)			12 Gauge		
29	CMP Uncoated Thickness		t	0.109	inch	2.8 mm
30	CMP Developed Width Factor		be	1.241		
31	<b>File Anchors</b>					
32	Number of Anchors		na	14		
33	Outer Anchors		no	14		
34	Inner Anchors		ni			
35	Outer Anchor Circle Diameter		Di	20.00	feet	
36	Inner Anchor Circle Diameter		Do		feet	
37	Effective Anchor Circle Diameter	$Da = (no * Do + ni * Di) / Do$	Da	20.00	feet	
38	Specified Anchor Post Tension Force		Pt	435	kips	
39	Anchor Bolt - Strength	Williams Grade 150, 2.5"	Fu	150,000	psi	
40	Anchor Bolt Area		ab	5.19	sq-in	
40	Anchor - Grout Strength		f'cg	3,000	psi	
2.0	<b>Loads</b> Based on: Preliminary Loading Data for MWT95/2.4-80m with IEC Class IIA					
	Base Shear	Characteristic Extreme loads	Fxy	117,665	lbs	523 KN
	Dead Load	no load factor	Fz	414,660	lbs	1,845 KN
	Maximum Moment	applied	Mxy	25,746,608	ft-lbs	34,908 KN-m
				P (kips)	V (kips)	M (k-ft)
	Dead Load		D	414.7		
	Extreme Wind		W		117.7	25,747 1.00
2.1	<b>Characteristic Loads</b>			414.7	158.6	34,758
2.3	<b>Applicable ACI Load Combinations - Section 9.2.1</b>			P <sub>u</sub> (kips)	V <sub>u</sub> (kips)	M <sub>u</sub> (k-ft)
	U = 0.9D+1.3W	Eq. (9-6) with exception (b) 1.3W		373.2	153.0	33,471 1.30
	U = 1.2D+1.3W	Eq. (9-5)		497.6	153.0	33,471 1.30
	U = 1.2D+1.0E	Eq. (9-7)		497.6	0.0	0 1.00
	IEC Load Case	IEC Load Factor		1.35	414.7	214.1 46,923



<b>3.0 Resistance To Base Shear (at Top of Foundation)</b>			
Allowable Shear	$F_v = 0.17 \cdot f_u \cdot a_b \cdot n_b$	$F_v =$	3,023 kips
Factored Base Shear		$V_u =$	153 kips
$F_v \gg V_u ?$			<b>OK</b>
<b>4.0 Required Thickness of Base Flange</b>			
Area of Base Flange	$A_F = \pi/4 [144 \cdot (O_d^2 - I_d^2) - n_b \cdot d^2]$		5,542 sq-in.
Compression at Base Flange	$f_{cbf} = (U_c + F_{pu}) \cdot n_b / A_F$	$f_{cbf} =$	3,605 psi
Required Thickness	$t_b = b_{fi} \cdot [3 \cdot f_{cbf} / (0.75 \cdot f_{yb})]^{0.5}$	$t_b =$	2.34 inches
Base Flange Thickness Design by Others		$B_{ft} =$	2.95 inches
$B_{ft} > t_b ?$			<b>OK</b>
<b>ACI 318 Code Compliance Checks</b>			
<b>5.0 Tower Anchor Bolt Analysis - ACI 318 Section 9.2.1</b>			
Dead Load	$D = F_z / n_b$	$D =$	2,962 lbs
Wind Load	$W = 4M_{ext} / (n_b \cdot D_{wall})$	$W =$	52,277 lbs
Earthquake	$E = 4M_{xy} / (n_b \cdot D_{wall})$	$E =$	0 lbs
Min. Required Post Tension Load	Set near $W - 0.9D$	$P =$	49,611 lbs
Specified Post Tension Load		$P_s =$	76,200 lbs
Effective Post Tension Load	$P_{se} = P_s \cdot 0.9$ with losses	$P_{se} =$	68,580 lbs
Factored Tension	$U_t = -0.9 D + 1.35 W$ (or 1.0 E)	$U_t =$	67,908 lbs
Allowable Anchor Tension	$F_b = a_b \cdot 0.7 f_{pu}$ per 18.5.1c	$F_b =$	88,900 lbs
$F_b > U_t ?$			<b>OK</b>
<b>5.5 Pile Anchor Bolt Analysis - ACI 318 Section 9.2.1</b>			
Foundation Weight	$W_{t-fnd} = \pi D_o^2 / 4 \cdot T \cdot \gamma'_c$	$W_{t-fnd} =$	339 kips
Dead Load	$D = (F_z + W_{t-fnd}) / n_a$	$D =$	54 kips
Wind Load	$W = 4M_{ext} / (n_a \cdot D_a)$	$W =$	368 kips
Earthquake	$E = 4M_{xy} / (n_a \cdot D_a)$	$E =$	0 kips
Min. Required Post Tension Load	Set near $W - 0.9D$	$P =$	319 kips
Specified Post Tension Load		$P_s =$	435 kips
Effective Post Tension Load	$P_{se} = P_s \cdot 0.9$ with losses	$P_{se} =$	392 kips
Factored Tension	$U_t = -0.9 D + 1.35 W$ (or 1.0 E)	$U_t =$	448 kips
Allowable Anchor Tension	$F_b = a_b \cdot 0.7 f_{pu}$ per 18.5.1c	$F_b =$	545 kips
$F_b > U_t ?$			<b>OK</b>
<b>6.0 Compression on Grout Bearing Analysis - ACI 318 Section 9.3.2.5</b>			
Factored Compression per bolt	$U_c = 1.2 D + 1.35 W$ (or 1.0 E)	$U_c =$	74,128 lbs
Area Base Flange	$A_F = \pi/4 [144 \cdot (O_d^2 - I_d^2) - n_b \cdot d^2]$		5,542 sq-in
Compression by moment	$C_m = U_c \cdot n_b / A_F$	$C_m =$	1,872 psi
Compression by post tension	$C_{ps} = P_s \cdot n_b / A_F$	$C_{ps} =$	1,925 psi
Total Compression on Grout		$C_{gt} =$	3,797 psi
Strength reduction factor		$\phi =$	0.85
Grout Compressive Strength		$f'_g =$	8,500 psi
Nominal Grout Strength	$C_{gn} = \phi \cdot f'_g$	$C_{gn} =$	7,225 psi
$C_{gn} > C_{ct} ?$			<b>OK</b>
<b>7.0 Concrete Stress under Grout Analysis - ACI 318 Section 9.3.2.5</b>			
Grout Trough Depth		$d_g =$	2.0 inches
Area under grout	$A_G = A_F + 2 \cdot d_g \cdot D_{wall} \cdot 12 \cdot \pi$	$A_G =$	7,664 sq-in
Compression	$C_{ct} = (U_c + P_s) \cdot n_b / A_G$	$C_{ct} =$	2,746 psi
Strength reduction factor		$\phi =$	0.85
Concrete Compressive Strength		$f'_c =$	6,000 psi
Nominal concrete strength	$C_{cn} = \phi \cdot f'_c$	$C_{cn} =$	5,100 psi
$C_{cn} > C_{ct} ?$			<b>OK</b>

8.0 Stresses In Concrete at Sustained Service Loads - ACI 318 - Section 18.4.2(a)			
<i>Grout under Tower Base Flange</i>			
Stress from Pre-Stress Force	$Sp = nb*Ps/A_F$	$Sp =$	1,925 psi
Stress from Sustained Tower Load	$Ss = Fz/A_F$	$Ss =$	75 psi
Total Stress on Grout	$St-s = Sp + Ss$	$St-s =$	2,000 psi
Nominal Grout Strength	$Sg = \phi*f_g, \phi = 0.45$	$Sg =$	3,825 psi
$Sg > St-s ?$			<b>OK</b>
<i>Concrete under Grout</i>			
Stress from Pre-Stress Force	$Sp = nb*Ps/A_G$	$Sp =$	1,392 psi
Stress from Sustained Tower Load	$Ss = Fz/A_G$	$Ss =$	54 psi
Total Stress on Concrete	$St-s = Sp + Ss$	$St-s =$	1,446 psi
Nominal Concrete Strength	$Sc = \phi*f_c, \phi = 0.45$	$Sc =$	2,700 psi
$Sc > St-s ?$			<b>OK</b>
9.0 Stresses In Concrete at Transient Service Loads - ACI 318 - Section 18.4.2(b)			
<i>Grout under Tower Base Flange</i>			
Service Load Moment (Msl)	Use Max Operational	$Msl =$	12,873 ft-kips
Axial Load from Msl per bolt	$Usl = 4Msl/n_b/D_{wall}$	$Usl =$	26,139 lbs
Stress from Moment	$Sm = nb*Usl/A_F$	$Sm =$	660 psi
Stress from Pre-Stress Force	$Sp = nb*Ps/A_F$	$Sp =$	1,925 psi
Stress from Sustained Tower Load	$Ss = Fz/A_F$	$Ss =$	75 psi
Total Stress on Grout	$St-t = Sm + Sp + Ss$	$St-t =$	2,660 psi
Nominal Grout Strength	$Sg = \phi*f_g, \phi = 0.60$	$Sg =$	5,100 psi
$Sg > St-t ?$			<b>OK</b>
<i>Concrete under Grout</i>			
Stress from Moment	$Sm = nb*Usl/A_G$	$Sm =$	477 psi
Stress from Pre-Stress Force	$Sp = nb*Ps/A_G$	$Sp =$	1,392 psi
Stress from Sustained Tower Load	$Ss = Fz/A_G$	$Ss =$	54 psi
Total Stress on Concrete	$St-t = Sm + Sp + Ss$	$St-t =$	1,923 psi
Nominal Concrete Strength	$Sc = \phi*f_c, \phi = 0.60$	$Sc =$	3,600 psi
$Sc > St-t ?$			<b>OK</b>
			17,454 KN-m
11.5 Shear Capacity Analysis - between Tower and Anchor ACI 318 - Section 11.3.1			
Beam Width per bolt pair	$bw = \pi*(Da+D_{wall})/n_b$	$bw =$	9.17 inches
Beam Depth	$d = T - 6'$	$d =$	54.00 inches
Shear Force Across Section	$Vu = 2*4*M_u/n_b/D_a$	$Vu =$	134.1 kips
Shear Stress	$v = Vu/(bw*d)$	$v =$	155 psi
Shear Capacity	Equation (11-3) $Vc = 2*f_c^{0.5}*bw*d$	$Vc =$	76.8 kips
Area of Stirrup Bar	<b>#10 bar</b>	$As =$	1.27 in <sup>2</sup>
Stirrup Spacing		$s =$	24.0 inches
Rebar Yield Strength		$f_y =$	60,000 psi
Nominal Shear Capacity	$Vs = 0.85As*f_y*d/s$	$Vs =$	145.7 kips
$\phi Vn > Vu?$	$Vn = \phi(Vc+Vs), \phi = 0.75$	$Vn =$	166.9 kips
			<b>OK</b>
<i>Therefore use #10 bar @ 24-inches or equivalent</i>			
11.7 Uplift Capacity Analysis of Embedment Ring			
Beam Width per bolt pair	$bw = \pi*D_{wall}/(n_b/2)$	$bw =$	7.58 inches
Eff. Depth for Embedment Ring	$d = T - 6' - 8'' - 3''$	$d =$	43.00 inches
Uplift (Shear) Force per Bolt	$Vu = Ut$	$Vu =$	67.9 kips
Shear Stress	$v = Vu/(bw*d)$	$v =$	208 psi
Shear (Uplift) Capacity of Half Cone projecting at 45 degrees upward	Equation (11-3) $Vc = 2*f_c^{0.5}*bw*d$	$Vc =$	50.5 kips
Area of Stirrup Bar	<b>#10 bar</b>	$As =$	1.27 in <sup>2</sup>
Stirrup Spacing	<b>Consider one bar through plane</b>	$s =$	24.0 inches
Rebar Yield Strength		$f_y =$	60,000 psi
Nominal Shear Capacity	$Vs = 0.85As*f_y*d/s$	$Vs =$	116.0 kips
$\phi Vn > Vu?$	$Vn = \phi(Vc+Vs), \phi = 0.75$	$Vn =$	124.9 kips
			<b>OK</b>
<i>Therefore use #10 bar @ 24-inches or equivalent</i>			

**11.9 Check Embedment Ring Thickness**

Embedment Plate Width		$b = b_f =$	10.94 inches
Embedment Plate Thickness		$t = P_t =$	1.50 inches
Required Thickness	$t_b = b_f \cdot [3 \cdot f_{cbf} / (f \cdot f_{yb})]^{0.5}$	$t_b =$	1.61 inches
			Marginally OK
Radius of between bolt pairs	$l = r_{bp} = \pi D_{wall} / n_b \cdot 2$	$l =$	7.58 inches
Consider embedment ring as cantilever beam between bolt pairs with compressive load set at $U_t$			
Compression at Base Flange	$f_{cbf} = U_t \cdot n_b / A_{BF}$	$f_{cbf} =$	1,715 psi
Compressive Load	$w = f_{cbf} \cdot b_f$	$w =$	18,770 lb/in
Moment	$M = w \cdot l^2 / 12$	$M =$	89,832 in-lbs
Section Modulus of Ring	$S = b \cdot d^2 / 3$	$S =$	8.21 in <sup>3</sup>
Bending Stress	$f_b = M / S$	$f_b =$	10,946 psi
Yield Strength Tower Base Plate	$\phi = 0.75$	$\phi f_{yb} =$	37,500 psi
$\phi f_{yb} > f_b ?$			<b>OK</b>

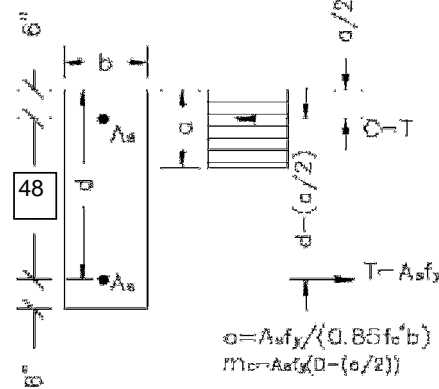
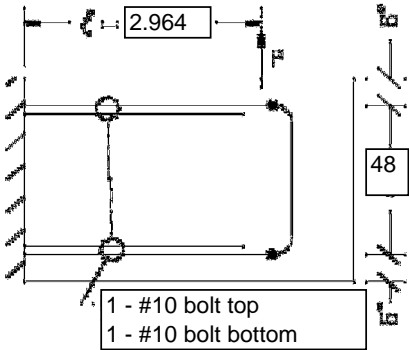
**12.0 Flexural Capacity - Reinforcement Analysis**

*Moment Calculation and Uplift - consider width of foundation at each bolt pair:*

Uplift Force Induced by Moment	$P_u = U_t \cdot \text{anchor} \cdot 2 \cdot n_a / n_b$	$P_u =$	89.6 kips
Beam Width	$b = \pi \cdot (D_a + D_{wall}) / n_b$	$b =$	0.765 feet
Uplift Force Arm	$L = (D_a - D_{wall}) / 2$	$L =$	2.964 feet
Moment Induced by Uplift Force	$M_p = P_u \cdot L$ (tension side)	$M_p =$	<b>265.6 ft-kips</b>
Elastic Section Modulus	$S = b d^2 / 6$	$S =$	5,505 in <sup>3</sup>
Elastic Bending Stress	$f_b = M / S$	$f_b =$	579 psi

*Section Check:*

*Double Reinforcement Check:*



**Design Radial Reinforcement - Top Section @ each bolt pair**

Area of Bar	1 - #10 bolt top	$A_s =$	1.27 in <sup>2</sup>
Bar Yield Strength		$f_y =$	60,000 psi
Bar Yield Force	$T = A_s f_y$	$T =$	76.2 kips
Location of Neutral Axis	$a = A_s f_y / (0.85 f'_c b)$	$a =$	1.63 inches
Bar Tension Force Arm	$d = T - 6"$	$d =$	54.0 inches
Nominal Moment Capacity	$\phi(A_s f_y)(d - a/2), \phi = 0.90$	$\phi M_c =$	304.0 ft-kips
$\phi M_c > M_p?$			<b>OK</b>

**Design Radial Reinforcement - Bottom Section @ each bolt pair**

Bearing Stress	$q = n_a \cdot P_t / A + P_u / (b \cdot L)$	$q =$	53.0 ksf
Unit Load	$w = q \cdot b$	$w =$	40.5 kips/ft
Moment Induced by Bearing	$M_q = w \cdot L^2 / 2$	$M_q =$	<b>178.0 ft-kips</b>
Elastic Bending Stress	$f_b = M / S$	$f_b =$	388 psi
Area of Bar	1 - #10 bolt bottom	$A_s =$	1.27 in <sup>2</sup>
Bar Yield Strength		$f_y =$	60,000 psi
Bar Yield Force	$T = A_s f_y$	$T =$	76.2 kips
Location of Neutral Axis	$a = A_s f_y / (0.85 f'_c b)$	$a =$	1.63 inches
Bar Tension Force Arm	$d = T - 6"$	$d =$	54.0 inches
Nominal Moment Capacity	$\phi(A_s f_y)(d - a/2), \phi = 0.90$	$\phi M_c =$	304.0 ft-kips
$\phi M_c > M_q?$			<b>OK</b>

**Design Circumferential Hoops**

Area of Rebar	1 - #9 rebar	$A_s =$	1.00 in <sup>2</sup>
Spacing (beam width)	@ 9 inches or equivalent	$s = b =$	0.75 feet
Rebar Yield Strength		$f_y =$	60,000 psi
Rebar Yield Force	$T = A_s f_y$	$T =$	60.0 kips
Elastic Section Modulus	$S = b d^2 / 6$	$S =$	5,400 in <sup>3</sup>
Elastic Bending Stress	$f_{by} = M / S$ take as 50% of $f_{bx}$	$f_b =$	290 psi
Max. Circumferential Bending Moment	$M_c = f_{by} \cdot S$	$M_c =$	130.3 ft-kips
Location of Neutral Axis	$a = A_s f_y / (0.85 f'_c b)$	$a =$	1.31 inches
Nominal Moment Capacity	$\phi(A_s f_y)(d - a/2), \phi = 0.90$	$\phi M_c =$	240.1 ft-kips
$\phi M_c > M_p?$			<b>OK</b>