

28 resistance is often crucial, as it mobilizes at much lower strains (or pile settlements)
29 than base resistance. There are also situations in which pile base resistance cannot
30 be relied upon, due to lack of quality of the pile base (e.g., debris accumulated at
31 bottom, or rock with cavities) (Seidel and Collingwood 2001).

32 Fundamentals of shear strength of interfaces have been investigated through
33 laboratory tests that lead to the development of analytical and empirical models to
34 predict the shear strength of rock-rock interfaces (Patton 1966; Barton and Choubey
35 1977) or of rock-concrete interfaces (Kodikara and Johnston 1994). Other models to
36 study the shear behavior of geotechnical interfaces considering their roughness
37 (Barton et al. 1985), or the influence of anisotropy and grain size of materials
38 involved in such interfaces (Kishida and Uesugi 1987; Wang et al. 2007) have also
39 been proposed. However, the side shear resistance of rock-socketed piles is usually
40 estimated using recommendations from codes and standards, or using local
41 knowledge obtained from full-scale static load tests performed in similar ground
42 (Seidel and Collingwood 2001); empirical correlations to estimate it as a function of
43 the uniaxial compressive strength (UCS or σ_c) are also common (Rezazadeh and
44 Eslani 2017) (Section 3.1.3 further discusses some of the most common ones).
45 However, in agreement with the conclusions of O'Neill et al. (1996) after their
46 analysis of 245 load tests in different materials, other parameters in addition to the
47 UCS are required for an improved estimation of the side shear resistance. As likely
48 candidates to be considered, the main factors affecting the side shear resistance
49 are: the construction method and drilling tools used (Nam and Vipulanandan 2008;
50 O'Neill et al. 1996; Ng et al. 2001), the type and quality of the rock mass (Haberfield

51 2013; Melentijevic and Olalla 2014), the depth and diameter of the socket (Seidel
52 and Haberfield 1995; Seol and Jeong 2007), the residual drilling fluid coating the
53 socket walls and the thickness of smear zone (Williams and Pells 1981; Haberfield
54 2013), and the initial normal stress and roughness at the pile-rock interface (Horvath
55 and Kenney 1979; Pells et al. 1980; Horvath et al. 1983; Seidel and Collingwood
56 2001; Seol et al. 2009).

57 This paper studies the influence of roughness at the socket's concrete-rock interface,
58 and how it affects rock-socketed piles. This is because (i) full-scale and model pile
59 tests demonstrate that socket roughness has a large influence on the side shear
60 resistance of rock-socketed piles (see Pells et al. 1980; Horvath et al. 1983; Seol
61 and Jeong 2007; Nam and Vipulanandan 2008; Dai et al. 2017); and because,
62 despite its importance, (ii) socket roughness is not often considered in current design
63 practice. To that end, the roughness factor (RF) defined by Horvath et al. (1983) is
64 used as a quantitative measure of roughness, and to investigate its influence on pile
65 performance. In particular, this paper applies the Distinct Element Method (DEM) to
66 model rock-socketed piles, and analyzes the effect of socket roughness on their side
67 shear behavior.

68 **2 Numerical model to simulate rock-socketed piles in DEM^{3D}**

69 **2.1 Introduction and fundamentals**

70 Particle Flow Code (PFC) (Itasca Consulting Group Inc 2014) is the commercial
71 software with a DEM implementation employed in this paper. In a DEM model, the
72 displacements and interactions of systems composed by rigid and finite-sized

73 particles –spherical in PFC^{3D}, see **Fig. 1(a)**– can be simulated. Mechanical
74 interactions are created or destroyed by the contact-detection algorithm whenever
75 bodies are sufficiently close or separated. The system evolution is computed through
76 explicit dynamic simulation schemes that apply Newton’s second law to the particles,
77 and that consider a force-displacement law at the contacts (Itasca Consulting Group
78 Inc 2014). PFC provides a Bonded-Particle Model (BPM) to simulate cohesive
79 material (e.g., concrete or rock) with several possible contact models (Potyondy
80 2015), as well as “interfaces” among materials to characterize their behavior when
81 they are in contact. In this work, the Flat-Joint Contact Model (FJCM) is employed to
82 simulate the pile and rock bodies, whereas the Smooth-Joint Contact Model (SJCM)
83 is employed to reproduce the behavior at the rock-concrete interface.

84 The FJCM simulates the contact between particles in the BPM using locally notional
85 (bonded or unbonded) surfaces (see **Fig. 1(a)**), which are discretized into elements
86 (Potyondy 2012). The SJCM simulates the macroscopic behavior of a bonded or
87 unbonded interface formed by several particles, in which particles can overlap or
88 slide past each other, to avoid moving around one another (see **Fig. 1(b)**). In both
89 models, the behavior of bonded interfaces is linear elastic until the bond strength is
90 reached and the bond breaks, making the interface unbonded (Itasca Consulting
91 Group Inc 2014). Unbonded interfaces have a linear, elastic and frictional behavior
92 in which the slip is simulated by a Coulomb limit on the shear force. For additional
93 details, see Potyondy (2015), Ivars et al. (2008), and Itasca Consulting Group Inc
94 (2014).

95 In this work, DEM numerical models are used to calculate the side shear resistance
96 of rock-socketed piles with different levels of socket roughness. To that end,
97 calibrating the micromechanical parameters that control the macroscopic behavior
98 of the rock and concrete, or of the rock-concrete interface, is needed first.
99 (Experimental data published by Gu et al. (2003) –about the shear behavior of
100 sandstone-concrete interfaces– and by Gutiérrez (2013) –about gneiss-concrete
101 interfaces– are used.) Finally, the procedure to generate samples for numerical
102 testing of rock-socketed piles is presented.

103 **2.2 Calibration of micromechanical parameters**

104 The calibration process has two steps: first, the FJCM parameters are calibrated,
105 using one UCS test as reference; then, the SJCM parameters are calibrated, using
106 one direct shear test as reference. Such calibration of micromechanical parameters
107 is conducted in 2D. The main reason for this is computational efficiency: 2D models
108 are much faster than 3D models, and many model runs are required for the
109 calibration. But, in agreement with previous research (see e.g., Ding et al., 2014),
110 results also indicate that the influence of the 2D vs. 3D calibration on the
111 macroscopic behavior of such tests is small; and we have also noted that the
112 behavior at the rock socket interface is similar to a (2D) direct shear test, with radial
113 horizontal displacements of particles and negligible particle rotations at the interface.
114 This suggests that the difference of pile behavior due to a 2D vs 3D calibration are
115 probably small and probably do not justify the large increase of computational cost
116 associated to a 3D calibration.

117 2.2.1 Calibration of the Flat-Joint Contact Model

118 To calibrate the micromechanical parameters of the FJCM employed for intact (rock
119 or concrete) materials, the uniaxial compressive strength (UCS) test is commonly
120 used to compare numerical and laboratory tests results. The procedure is iterative
121 and starts matching the macroscopic Poisson's ratio (ν), which mainly depends on
122 the particle and flat-joint normal-to-shear stiffness ratio (k^* and $\overline{k^*}$); next, the
123 macroscopic Young's modulus (E), is calibrated modifying the particles and flat-joint
124 effective modulus (E^* and $\overline{E^*}$); finally, the UCS (σ_c) is adjusted by tuning the cohesion
125 (c) and tensile strength (σ_t) of the flat-joint. There are others parameters of the FJCM
126 –such as the friction angle of the material (ϕ), the bonded fraction ($\phi_B = \eta_B/\eta_{FJ}$,
127 where η_B and η_{FJ} are the number of bonded and total flat-joint contacts,
128 respectively), the initial surface gap (g_o), and the gapped fraction ($\phi_G = \eta_G/\eta_{FJ}$,
129 where η_G is the number of gapped contacts: i.e., contacts between particles with a
130 separation greater than g_o , so there are not interactions between particles),– that
131 could also influence the behavior of the material, but they are not considered in the
132 calibration conducted herein (for details, see Gutiérrez-Ch et al. 2018).

133 **Table 1** lists the micromechanical properties obtained with numerical samples of
134 sandstone (types S2 and S3), gneiss and concrete (types C1 and C2) after the
135 calibration process. **Table 2** compares the macroscopic UCS test results – σ_c, E, ν –
136 obtained experimentally by Gutiérrez (2013) (for gneiss and C2-concrete) and by Gu
137 et al. (2003) (for S2- and S3-sandstone and C1-concrete) with those simulated with
138 DEM^{2D} employing the parameters listed in **Table 1**. It can be observed that (i) the
139 average values calculated with the DEM^{2D} numerical models agree well with the

140 laboratory results, and (ii) that the standard deviation of results obtained in
141 experimental tests (when reported) is greater for rock samples than for concrete.
142 This is probably due to the presence of minor defects in rock samples –e.g., weak
143 planes, microcracks, etc.– which are probably not so common in a manufactured
144 material such as concrete. Note also that the DEM results present similar standard
145 deviation for rock and concrete samples, as the sample generation algorithm is
146 equivalent for both.

147 *2.2.2 Calibration of the Smooth-Joint Contact Model*

148 The SJCM parameters –i.e., the smooth-joint’s normal and shear stiffness ($k_{n\ SJ}$ and
149 $k_{s\ SJ}$); and its coefficient of friction ($\mu_{\ SJ}$)– are calibrated comparing numerical and
150 experimental results of direct shear tests on unbonded (i.e., with $c = 0$) rock-
151 concrete planar interfaces under different normal stresses. To that end, numerical
152 specimens are generated first, using the Shear Box Genesis (SBG) method
153 proposed by Bahaaddini et al. (2013). **Fig. 2(a)** shows an example of one of such
154 specimens formed by an interface between concrete-C1 and sandstone-S2 (the
155 interface is modelled using the smooth-joint contact model, and the interactions
156 between particles of each body are modelled using the flat-joint contact model with
157 the micro-mechanical properties listed in **Table 1**).

158 Then, the numerical joint normal stiffness ($K_{n\ DEM}$) is calibrated against the laboratory
159 normal stiffness ($K_{n\ lab}$) through a normal deformability test on planar joints, in which
160 the value of $k_{n\ SJ}$ is adjusted. Finally, the joint shear stiffness ($K_{s\ DEM}$) and the
161 coefficient of friction (μ_{DEM}) are calibrated by trial and error, adjusting the value of
162 $k_{s\ SJ}$ and $\mu_{\ SJ}$ to reproduce the behavior of laboratory shear tests (i.e., the shear

163 stiffness, $K_{S\ lab}$, and the peak shear strength measured in the laboratory). The
164 DEM^{2D} results of direct shear tests on unbonded sandstone (S2)-concrete (C1)
165 planar interfaces are presented in **Fig. 2(b)**, and the calibrated SJCM parameters
166 are listed in **Table 3**. Results show that DEM^{2D} models can reproduce the behavior
167 of concrete-rock interfaces, so that they agree well with experimental results
168 reported by Gu et al. (2003) and by Gutiérrez (2013) (see **Table 4Table 3**). For
169 additional details about calibration, see also Bahaaddini et al. (2013) and Gutiérrez-
170 Ch et al. (2018).

171 **2.3 Generation of samples to simulate rock-socketed piles**

172 **Fig. 3** presents an idealized sub-surface profile of a rock socket, and the associated
173 DEM^{3D} model developed to represent it. The methodology to simulate rock-socketed
174 piles in this DEM^{3D} model can be summarized as follows:

175 *2.3.1 Initial particle assembly of rock and pile bodies*

176 The 3D model of the rock-socketed pile consists of several frictionless walls to avoid
177 shear forces from developing at ball-wall boundary contacts and to reduce boundary
178 effects. Such walls are filled with two assemblies of randomly placed particles that
179 constitute the concrete pile and the rock around it (**Fig. 4(a)**). The normal stiffness
180 of the walls is set to be larger than the average particle normal stiffness to ensure
181 that particle–wall overlaps remain small. Particle sizes at the different zones
182 considered follow a uniform distribution of radii between R_{min} and R_{max} bounds
183 chosen after a sensitivity analysis (not reproduced herein) that indicated that socket
184 behavior is not affected by particles whose size (at zone 1) are within this interval
185 (see **Table 5**).

186 To optimize the computational cost, (i) the rock body was discretized into three
187 zones, defined as a function of the nominal socket radius, R , and with the smallest
188 particles being positioned adjacent to the pile-rock interface (zone 1) (see Fig. 4(a));
189 and (ii) a portion of only 1/8 of the pile cross-section (or of a 45-degree angle) was
190 modeled, instead of the whole cross section (see Fig. 4(b)). The base was made
191 void because only the pile shaft behavior is analyzed; note also that the particles
192 have a strong cohesion, which prevents them from falling into the void. The nominal
193 socket radius (R) and the nominal socket length (L) are 0.4 m and 0.8 m,
194 respectively, in all analyses presented herein.

195 *2.3.2 Application of an isotropic initial stress*

196 After the initial assembly of particles is created, the “secondary walls” employed to
197 discretize the rock into three zones are removed (see **Fig. 4(a)**). Then, to get a better
198 distribution of contacts and to reduce the locked-in forces, the radii of all particles
199 are iteratively changed until a target isotropic stress ($\sigma_o^c \cong 1\%$ of the uniaxial
200 compressive strength) is reached. To do this, “measurement regions” –spheres in
201 3D– are installed inside the pile and rock bodies, and the isotropic stress ($\sigma_o =$
202 $(\sigma_{11} + \sigma_{22} + \sigma_{33})/3$) within each “measurement region” is computed; see **Fig. 4(c)**.
203 The process finishes when the normalized difference between the target isotropic
204 stress in all spheres and the specified value of isotropic stress is less than the
205 tolerance (in agreement with Bahaaddini et al. 2013 and Gutiérrez-Ch et al. 2018,
206 the following relation is used $\frac{\sigma_o^c - \sigma_o}{\sigma_o^c} \leq 0.5$).

207 *2.3.3 Elimination of floating particles*

208 “Floating” particles with less than three contacts can appear during the previous step,
209 as illustrated by the red balls shown in **Fig. 4(d)**. When this occurs, the radii of
210 floating particles are increased until all particles away from the specimen boundaries
211 have at least three contacts.

212 *2.3.4 Application of Flat-Joint and Smooth-Joint Contact Models*

213 First, flat-joint contacts are installed between particles in each material with a gap
214 lower than g_o (the micromechanical parameters of such flat-joint contacts are listed
215 in **Table 1**). Next, the wall corresponding to the intended contact –the rock-concrete
216 interface– is removed. Finally, particles in contact with the intended rock-concrete
217 surface are identified, and the SJCM is applied to the contacts between balls forming
218 the rock and concrete bodies, employing the micromechanical properties listed in
219 **Table 3**.

220 **Fig. 5(a)** shows the model after these contact models are applied; it also shows –
221 using purple arrows that represent the unit normal vectors (\hat{n}_j) of the SJCM applied
222 at each contact at the rock-concrete interface (see **Fig. 1(b)**)– how the socket
223 roughness affects the orientation of such contacts. To do that, and since the
224 orientations of the interface contacts varies depending on the socket roughness, a
225 FISH function –FISH is an embedded scripting language that enables the user to
226 interact with, and to manipulate, PFC models (Itasca Consulting Group Inc 2014)–
227 was developed to define and update the unit normal vector (\hat{n}_j) of the SJCM
228 perpendicular to the rock-concrete interface. Such unit normal vectors have been

229 plotted in stereographical projections (Ragan 2009) that illustrate the variability of
230 orientations due to roughness, as shown in **Fig. 5(b)**.

231 **Fig. 6** illustrates the shapes of the rock-socketed piles considered in this work, and
232 their boundary conditions; it also shows how the DEM^{3D} model of a portion of the
233 rock socket is created, with the corresponding pile self-weight and overburden loads
234 (Q_1 and Q_2 , respectively) acting on it. 16 rock-socketed piles (6 on sandstone-S3, 5
235 on sandstone-S2, and 5 on gneiss) with different degrees of socket roughness were
236 modelled. The RF is employed to quantify roughness; it is defined as (Horvath et al.
237 1983):

$$RF = \frac{h_m L_t}{RL} \quad (1)$$

238 where h_m is the average height of asperities, R is the nominal socket radius, L_t is
239 the total travel distance along the socket wall, and L is the nominal socket length
240 (see **Fig. 7**).

241 According to O'Neill et al. (1996) and Hassan et al. (1997), the socket roughness
242 was simulated using sinusoidal surfaces with amplitudes of 0, 4, 10, 20, 32.5 and 40
243 mm; and with a wavelength of 250 mm. These values correspond to roughness
244 factors of $RF = 0.000$, $RF = 0.010$, $RF = 0.025$, $RF = 0.050$, $RF = 0.085$, $RF =$
245 0.106 . **Fig. 6(b)** illustrates the roughness profiles associated to each RF , using radial
246 cross-sections of the corresponding DEM^{3D} models. Also remember that (i) the
247 micro-properties used for the FJCM and the SJCM are listed in **Table 1** and **Table**
248 **3** (note that, although the calibrated SJCM parameters used in DEM^{3D} models
249 originate from DEM^{2D} models, its expected influence on the obtained results is small,

250 see e.g., Ding et al., 2014), and (ii) that the particle size distribution (R_{max}/R_{min}) is
251 presented in **Table 5**. The reader should note that a “perfectly smooth” interface with
252 a $RF = 0.000$ might not actually be achievable in practice with any of the usual drilling
253 tools. However, numerical models with $RF = 0.000$ are also considered to provide a
254 lower bound.

255 *2.3.5 Initialization of in situ stress and boundary conditions*

256 The next steps, as shown in **Fig. 6(a)**, are to apply a gravity load to induce in situ
257 stresses due to self-weight, and to apply external stresses that represent the
258 overlying loads due to the soil stratum (Q_2) and to the self-weight of the upper portion
259 of the pile (Q_1). To that end, a servo-control mechanism is employed, as it enables
260 one to control the translational velocity of selected walls, so that a desired force can
261 be applied or maintained on them (Itasca Consulting Group Inc 2014). This allows
262 one to model only the rock socket, hence significantly reducing the size of the
263 computational model, and the associated computational cost. Note, therefore, that
264 the analyses herein neglect the elastic shortening of the portion of the pile above the
265 rock socket, as well as the (probably small) side shear resistance developed in it.

266 To conduct the numerical pile tests, the displacements of the boundary walls or fixed
267 walls shown in **Fig. 6(a)** are restraint, and an axial velocity (V_z) of 0.05 m/s –or low
268 enough to ensure that the rock-concrete specimens remain in quasi-static
269 equilibrium– is applied on the moving wall that represents the rock-socket head. The
270 axial force (P) and the settlement of the socket head (δ) are obtained, respectively,
271 as the force reaction recorded at the wall representing the rock-socket head and as
272 the axial displacements of such wall.

273 3 Results

274 3.1.1 Load-settlement response

275 The load-settlement curves from the DEM^{3D} simulations of six piles socketed into
276 sandstone-S3 (with different roughness profiles) are shown in **Fig. 8**. (Remember
277 that side loads developed along the portion of the pile embedded in soil, as well as
278 its elastic shortening, are neglected, see **Fig. 3**). All analyses were conducted until
279 a limit socket head settlement (δ) of 10% of the socket diameter ($0.1D$) is reached.
280 Results in **Fig. 8** show that socket roughness is an important factor that significantly
281 increases the load capacity and stiffness of rock-socketed piles: for instance, and as
282 shown in **Table 6**, the model with $RF = 0.050$ supported a working load 2.1 times
283 higher than that supported by the model with $RF = 0.025$, for an identical pile head
284 settlement of $0.02D$. (A similar behavior is also obtained in piles socketed into
285 sandstone-S2 and gneiss). This qualitatively agrees with the results of field tests
286 conducted on sandstone by Pells et al. (1980), in which the “rough” piles supported
287 a higher working load –about 1.2 to 1.7 times higher, depending on the UCS of the
288 rock (sandstone with $\sigma_c = 6 - 40$ MPa)– than the “smooth” piles. Similarly, an
289 influence of RF on the load test results of pile sockets in shale was also observed by
290 Horvath et al. (1983), who reported that the pile with $RF = 0.076$ supported a working
291 load 1.3 times higher than the model with $RF = 0.036$, for a pile head settlement
292 equivalent to $0.02D$, see **Table 6**.

293 3.1.2 Average side shear resistance

294 Considering that DEM^{3D} models are composed by an assembly of particles, and that
295 the pile models considered herein have a void at their base, the average side shear

296 resistance developed along the socket was obtained through an averaging
297 procedure, as:

$$f_{ave} = \frac{\sum F_z}{A} \quad (2)$$

298 where f_{ave} is the average side shear resistance (for a given settlement), $\sum F_z$ is the
299 sum of all vertical components of contact forces acting on all particles at the rock-
300 concrete interface, and A is the nominal side area of the rock-concrete interface.

301 The computed curves relating f_{ave} and δ for rock sockets with different roughness
302 are presented in **Fig. 9(a)** and **Fig. 10**. Results show that the average side shear
303 resistance increases as the roughness of the interface increases; for instance, f_{ave}
304 of the models with $RF = 0.025$ is about 2 times higher than that obtained with $RF =$
305 0.010 . This is because the rougher interface (i.e., with higher RF), tends to dilate
306 more and, due to the constraints at the rock socket, which is probably close to a
307 constant normal stiffness (CNS) boundary condition (see Gutierrez-Ch et al. 2018
308 for a discussion), this significantly increases the normal stresses at the rock-socket,
309 hence significantly increasing the side shear resistance of interfaces with higher RF
310 (see e.g., Seol and Jong 2007; Gutierrez-Ch et al. 2018).

311 The analysis of the number of micro-cracks developed at the body of the pile, and at
312 the rock surrounding it, also provides interesting information. **Fig. 9(b)** shows that
313 the number of such microcracks in piles socketed into sandstone-S3 is very small
314 before reaching a settlement of $0.01D$, and that it increases, with a higher rate for
315 rougher sockets, after such threshold. (Similar results are also obtained in
316 sandstone-S2 and gneiss). This finding is supported by previous works (see e.g.,

317 Bahaaddini et al. 2013; Gutiérrez-Ch et al. 2018), suggesting that models with lower
318 RF values fail through sliding along the rock-concrete interface, whereas the
319 degradation of asperities becomes more relevant for higher RF values (the need to
320 break the asperities is also responsible for the higher side shear resistances
321 developed along rougher interfaces). Note also that the number of cracks (due to
322 shear and tension) is greater in the rock than in the pile. This was expected, as the
323 sandstone rock considered in these models is much weaker than the concrete
324 forming the piles. Results in **Fig. 9(b)** suggest that the criterion of considering a
325 maximum settlement of $0.01D$ under design working loads (see e.g., Whitaker and
326 Cooke 1966; Horvath and Kenney 1979) is probably appropriate to avoid excessive
327 damage of rock-concrete interface of rock socket piles.

328 **Fig. 11** shows the relationship between the roughness of the rock-concrete interface
329 (as defined by the RF roughness factor) and its average side resistance, for two
330 threshold values of pile head displacements. The DEM^{3D} results are computed for
331 socket head settlements of 1% and 2% of the pile diameter, and they suggest that
332 the relationship is approximately linear for lower settlements, for which a significant
333 degradation of the socket interface has not yet occurred. However, the relationship
334 seems to be non-linear for higher settlements, associated to a significantly higher
335 number of cracks (see **Fig. 9(b)**). In agreement with previous research (see e.g., Dai
336 et al. 2017), this suggests that there might be an upper roughness limit, so that the
337 side resistance stops increasing (or increases much slower) when the roughness
338 increases beyond such threshold. In any case, results in **Fig. 11** confirm that socket
339 roughness is an important factor that could be incorporated in the design of rock-

340 socketed piles. Next, the authors present additional DEM^{3D} analyses, comparing
341 their results with the results of several other methods proposed in the literature,
342 some of which consider the socket roughness for their predictions.

343 3.1.3 Comparisons with other design methods

344 This section compares the results of the numerical pile tests conducted with DEM^{3D}
345 –in particular the average side shear resistance for socket head settlements of
346 $0.01D$, or $f_{ave,0.01D}$ – with estimations of other empirical and analytical criteria
347 published in the literature that also used $0.01D$ as a reference settlement to estimate
348 average side shear resistance.

349 To that end, the range of socket roughness that can be expected in real projects is
350 analyzed first, based on a literature review conducted by the authors. The results of
351 this analysis are compiled in **Table 7**. As it can be observed, and as supported by
352 previous works (see e.g., Collingwood 2000; Seidel and Collingwood 2001), socket
353 roughness is mainly a function of the drilling tools employed and of the type and
354 strength of the rock. In particular, **Table 7** suggests that sockets drilled with standard
355 tools in soft to medium rock tend to be relatively smooth (with height of asperities
356 less than 10 mm), unless the rock is heavily fractured or a special tool is used to
357 increase their roughness.

358 Empirical criteria that estimate the side shear resistance of the rock socket as a
359 function of the uniaxial compressive strength are employed, as follows:

$$f_{ave,peak}[\text{MPa}] = \alpha \sigma_c[\text{MPa}]^\beta \quad (3)$$

360 where $f_{ave,peak}$ is the average ultimate side shear resistance, σ_c is the intact UCS of
361 the weaker of the materials involved at the shear surface (rock or pile), and α and β
362 are empirical factors that depend on the type of rock, or in some cases on the
363 roughness (Seo et al. 2013). (Other authors –see e.g., Kulkarni and Dewaikar 2016–
364 have also proposed correlations with the UCS of the rock mass; such correlations,
365 however, are not reviewed herein).

366 **Table 8** shows compilation of methods that conform to Equation (3), and that have
367 been proposed in the literature to estimate socket side resistance. (Note that, some
368 of these methods incorporate the influence of roughness, using the roughness
369 classes defined by Pells et al. (1980) that are listed in **Table 9**).

370 Results computed herein are also compared with the results of other methods
371 proposed in the literature to estimate the average side shear resistance considering
372 the UCS of the material and the socket roughness (see e.g., Pells et al. 1980;
373 Horvath et al. 1983) as well as with the results of the method by Seidel and
374 Collingwood (2001), which considers additional parameters –e.g., the effect of
375 construction method (η_c), the Poisson's ratio (ν), etc.– through a non-dimensional
376 factor called the shaft resistance coefficient (SRC).

377 **Fig. 12** shows the comparison between these analytical and empirical solutions with
378 our DEM^{3D} results; **Fig. 12** compares (i) the average side shear resistance computed
379 with the Sandstone-S3 DEM^{3D} model for a head settlement of $0.01D$, $f_{ave,0.01D}$, and
380 (ii) its predictions computed with the empirical and analytical methods mentioned
381 above. Unless it is specified that the method is valid for certain types of socket only
382 –e.g., smooth or rough–, all the RF values are considered for the DEM^{3D} simulations.

383 This is why, for methods that not consider socket roughness, $f_{ave,0.01D}$ results are
384 same regardless of RF . Results in **Fig. 12** suggest that, despite the uncertainties
385 involved, the DEM^{3D} model is able to reproduce the overall predictions of the
386 methods analyzed, when an adequate range of roughness (or of RF values) is
387 considered. In that sense, for instance, results with the formulations by Serrano et
388 al. (2015) –which is developed for “smooth” piles– and by Rosenberg and Journeaux
389 (1976) –which employed data of load tests conducted on artificially roughened
390 sockets reported by Matich and Kozicki (1967), as well as sockets constructed in
391 highly fractured rock, which probably produced rough walls– agree quite well with
392 our DEM^{3D} results for $RF = 0.010$ ($h_m = 4$ mm) and for $RF = 0.106$ ($h_m = 40$ mm),
393 respectively; which are therefore within the “smooth” range of real piles or the “rough”
394 range of artificially roughened piles (see **Table 7**).

395 The methods of Williams et al. (1980) and Horvath and Kenny (1979) also provide
396 $f_{ave,0.01D}$ results that are similar to those computed with DEM^{3D} for relatively smooth
397 sockets (see **Fig. 12**); this agrees with the observation in **Table 7**, as (i) Williams et
398 al. (1980) piles were mainly constructed in a weak mudstone that produced
399 asperities with low height (with $h_m \leq 10$ mm calculated using socket roughness
400 profiles, see Williams 1980), and (ii) Horvath and Kenny (1979) indicate that they
401 worked with “load-test data concerning rock-socketed piers constructed by
402 conventional procedures”, and based on the information provided in **Table 7**, this
403 suggests that the average height of asperities is probably less than 10 mm. A similar
404 discussion can be made about the results with the method proposed by Arioglu et
405 al. (2018): although the authors do not report details about the drilling tools

406 employed, the information provided about rock types and rock strengths suggests
407 that average roughness were probably less than 10 mm, since standard drilling tools
408 were used (see **Table 7**).

409 Out of the formulations that specifically consider the socket roughness, the criterion
410 of Rowe and Armitage (1987) –which was proposed using loads test for which socket
411 roughness were measured ($h_m \leq 75$ mm)– provides values similar to the “smoother”
412 DEM^{3D} models (with $RF \leq 0.050$; see **Fig. 12**), whereas numerical results are
413 significantly higher than the predictions of Rowe and Armitage (1987) for rougher
414 sockets. This was also expected, as Rowe and Armitage (1987) acknowledge that
415 “[their method] may be quite conservative for artificially roughened sockets”. The
416 methods of O’Neill and Reese (1999), Canadian Foundation Engineering Manual
417 (2006) and AASHTO (2008) also provide similar results to the DEM^{3D} models with
418 $RF < 0.050$, while they are again conservative for higher RF values (see **Fig. 12**).
419 Again, this can be explained by the fact that these formulations are based on the
420 recommendations by Horvath and Kenny (1979) and Rowe and Armitage (1987).
421 Finally, DEM^{3D} results in this study provide $f_{ave,0.01D}$ estimates that are significantly
422 lower than those provided by the formulation by Seidel and Collingwood (2001). The
423 reasons for this discrepancy, or why Seidel and Collingwood (2001) predict
424 significantly higher $f_{ave,0.01D}$ values than other authors, are not clear.

425 Finally, **Fig. 13** compares, for sockets constructed in sandstone (S2, S3) and gneiss,
426 the (i) $\alpha = f_{ave,0.01D}/\sigma_c$ values obtained using the DEM^{3D} model, with (ii) load test
427 data and curves proposed by Pells et al. (1980) (see **Fig. 13(a)**), by Williams and
428 Pells (1981) (see **Fig. 13(b)**), and by Seidel and Collingwood (2001), after they

429 considered the load tests compiled by Kulhawy and Phoon (1993) (see **Fig. 13(c)**).
430 (Note that these α values correspond to a special case of Eq. (3) when $\beta = 1$;
431 however, α is used, since α is the usual symbol to relate socket side shear resistance
432 and UCS). It can be noted that the DEM^{3D} numerical results: (i) are in a good
433 agreement with α values reported by Pells et al. (1980) (note, however, that their
434 design curves do not capture the variability measured in the load tests), (ii)
435 reproduce well the range of responses measured in real pile reported in curves and
436 data proposed by Williams and Pells (1981) even for several rocks with different
437 strengths within the medium strength range, and (iii) are again consistent with the
438 data compiled by Kulhawy and Phoon (1993) for sockets with different roughness
439 constructed in medium strength rock, and with the design curves proposed by Seidel
440 and Collingwood (2001) to consider the influence of socket roughness.

441 In any case, and although the discussion above is somewhat speculative due to the
442 ever-existing difficulties to reproduce the real field behavior, and to interpret field
443 tests conducted by others, it strongly suggests that, when properly calibrated, the
444 DEM^{3D} models employed herein can be an adequate tool to estimate the influence
445 of the socket roughness on the average shaft resistance of real piles.

446 **4 Conclusions**

447 The side shear resistance of rock-sockets heavily depends on the roughness of the
448 socket interface. Many empirical criteria developed to assess the side shear
449 resistance of rock sockets, however, do not consider the influence of roughness,
450 providing estimates that only depend on the uniaxial compressive strength (UCS) of
451 the weaker material at the rock-concrete interface. Although there are also empirical

452 criteria that consider the socket roughness, they often provide widely variable
453 estimations of side shear resistance, especially for sockets with higher roughness.

454 This paper shows that discrete numerical models constructed with DEM^{3D} can be
455 employed to (i) reproduce the load-settlement behavior of rock sockets with different
456 roughness and (ii) to estimate their side shear resistance. In particular, results show
457 that DEM^{3D} is able to reproduce the range of load test results measured in practice
458 (or obtained analytically, when analytical solutions are available) for sockets with
459 similar roughness values, and constructed in rocks with UCS values in the medium
460 strength range of, approximately, 10–100 MPa. Results of this research also suggest
461 that sockets drilled with standard tools in soft to medium rock tend to be relatively
462 smooth, unless they are artificially roughened with special tools, and that the
463 commonly employed settlement threshold of 1% of socket diameter could represent
464 the settlement value beyond which significant degradation of asperities at the rock-
465 concrete interface begins to occur, especially for sockets with greater roughness.

466 The main advantage of the numerical pile load-tests employed herein is that they
467 are cheaper and faster than field tests, as they are relatively easy to calibrate, using
468 only laboratory tests results from (i) uniaxial compression strength tests and (ii) direct
469 shear tests under constant normal stiffness boundary conditions. They also allow a
470 more detailed quantification of the influence of roughness than many empirical
471 methods developed using field test data.

472 5 Data Availability Statement

473 Some or all data, models, or code that support the findings of this study are available
474 from the corresponding author upon reasonable request (scanned test data; analysis
475 results).

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Table 1. Micro-mechanical properties fitted from the UCS tests employed in the calibration of intact material models

Type	Property	(I)	(II)	(III)	(IV)	(V)
Particle micromechanical properties	E^* (GPa)	1.70	1.90	27.00	43.00	26.20
	$k^* = k_n/k_s$	1.35	1.45	2.75	3.25	3.00
	Friction angle ϕ ($^\circ$)	32	35	30	35	30
	Ball density, ρ (kg/m^3)	2500	2550	2500	2690	2500
	Minimum radius, R_{min} (mm)	1.0	1.0	0.8	0.8	1.0
	R_{max}/R_{min}	1.4	1.4	1.5	1.5	1.4
Flat-joint micromechanical properties	\bar{E}^* (GPa)	1.70	1.90	27.00	43.00	26.20
	\bar{k}^*	1.35	1.45	2.75	3.25	3.00
	c (MPa)	4.85	7.90	13.55	40.80	18.80
	σ_t (MPa)	2.8	3.5	6.0	20.0	9.0
	FJ bonding ratio, c/σ_t	1.73	2.26	2.26	2.04	2.08
	Bonded fraction, ϕ_B	0.88	0.90	0.93	0.96	0.94
	Gapped fraction, ϕ_G	0.12	0.10	0.07	0.04	0.06
	Initial gap, g_o (mm)	0.05	0.05	0.05	0.05	0.05

(I) Sandstone-S2, (II) sandstone-S3, (III) concrete-C1; (IV) gneiss, (V) concrete-C2

Table 2. Macro-mechanical properties for the UCS tests employed in the calibration of intact material models

Sample	Macro-properties	Experimental	PFC ^{2D}
		Ave.	Ave.
Gneiss [*]	σ_c (MPa) \pm SD	158.87 \pm 7.11 (<i>n</i> = 3)	158.84 \pm 7.14 (<i>n</i> = 10)
	E (GPa) \pm SD	50 [*]	50.23 \pm 0.40 (<i>n</i> = 10)
	ν \pm SD	0.25 [*]	0.24 \pm 0.03 (<i>n</i> = 10)
Concrete-C2 [*]	σ_c (MPa) \pm SD	62.20 \pm 0.74 (<i>n</i> = 4)	62.12 \pm 3.01 (<i>n</i> = 10)
	E (GPa) \pm SD	30 [*]	29.98 \pm 0.26 (<i>n</i> = 10)
	ν \pm SD	0.20 [*]	0.20 \pm 0.02 (<i>n</i> = 10)
Sandstone-S2 ^{**}	σ_c (MPa) \pm SD	11.49 \pm 3.19 (<i>n</i> = 8)	11.59 \pm 0.97 (<i>n</i> = 10)
	E (GPa) \pm SD	1.98 \pm 0.68 (<i>n</i> = 5)	2.01 \pm 0.02 (<i>n</i> = 10)
	ν \pm SD	0.10 \pm 0.00 (<i>n</i> = 3)	0.10 \pm 0.01 (<i>n</i> = 10)
Sandstone-S3 ^{**}	σ_c (MPa) \pm SD	21.77 \pm 1.19 (<i>n</i> = 8)	21.65 \pm 1.43 (<i>n</i> = 10)
	E (GPa) \pm SD	3.25 \pm 0.39 (<i>n</i> = 5)	2.32 \pm 0.02 (<i>n</i> = 10)
	ν \pm SD	0.10 \pm 0.00 (<i>n</i> = 3)	0.11 \pm 0.01 (<i>n</i> = 10)
Concrete-C1 ^{**}	σ_c (MPa) \pm SD	40 \pm 0.33 (<i>n</i> = 3)	39.87 \pm 2.73 (<i>n</i> = 10)
	E (GPa) \pm SD	29.95 \pm 0.08 (<i>n</i> = 3)	30.08 \pm 0.23 (<i>n</i> = 10)
	ν \pm SD	0.20 [*]	0.20 \pm 0.02 (<i>n</i> = 10)

SD = Standard Deviation, *n* = number of tests, *SD not reported in the original reference
^{*} Experimental from Gu et al. (2003), ^{**} Experimental from Gutiérrez (2013)

Table 3. Fitted parameters (with DEM^{2D} models) of SJCM on planar interface employed for validation

SJCM parameters (PFC ^{2D})	Sandstone(S2)- concrete(C1)	Sandstone(S3)- concrete(C1)	Gneiss- concrete(C2)
Joint normal stiffness, $k_{n\ SJ}$ (MPa/mm)	10	10	10
Joint shear stiffness, $k_{s\ SJ}$ (MPa/mm)	5	2	2
Joint coefficient of friction, μ_{SJ} ($\tan \phi$ (°))	0.70	0.70	0.60

Table 4. Comparison of joint macroscopic parameters for direct shear tests with DEM^{2D} results and measured in the laboratory by Gu et al. (2003) and Gutiérrez (2013)

Joint macroscopic parameters	Experimental from Gu et al. (2003)	Numerical (PFC ^{2D})	Experimental from Gu et al. (2003)	Numerical (PFC ^{2D})	Experimental from Gutiérrez (2013)	Numerical (PFC ^{2D})
System normal stiffness, K_n , (MPa/mm)	-	9.50	-	10.30	-	16.49
System shear stiffness, K_s , (MPa/mm)	1.84	1.81	1.37	1.36	2.51	2.53
Friction angle, $\tan \phi$ (°)	35.37	34.99	34.96	34.99	31.22	31.10

Table 5. Particle size distributions in the numerical models of rock-socketed piles testing

Material	Zone	Minimum radius, R_{min} (cm)	R_{max}/R_{min}
Rock	Zone 1	1.00	1.5
(Sandstone(S2,S3)	Zone 2	1.50	1.5
and gneiss)	Zone 3	2.25	1.5
Pile			
(Concrete(C1,C2))	Zone 1	1.00	1.5

Table 6. Axial load supported by rock-socketed piles with different roughness for socket head settlement of 2 % of pile diameter

Tests	Model	Roughness classification*	Roughness Factor, RF	Axial Load, P (kN)
Numerical PFC ^{3D} on sandstone-S3	1	R1	0.000	20.2
	2	R2	0.010	148.7
	3	R3	0.025	451.7
	4	R4	0.050	941.1
	5	R4	0.085	1675.4
	6	R4	0.106	2438.9
Field tests on shale by Horvath et al. (1983)	P1	R4	0.036	3996.5
	P2	R4	0.076	5189.3
	P3	R4	0.100	7273.9

* Roughness classification proposed by Pells et al. (1980), see **Table 9**

Table 7. Socket roughness resulting by different drilling method on rock-socketed piles

Drilling tool	Average height of asperities, h_m , (mm)	Type of rock	σ_c (MPa)	Reference of load test data
Auger ^b	75 ^c	Shale	0.48	Matich and Kozicki 1967
Diamond tube	< 4 ^e	Sandstone	14	Pells et al. 1980
Diamond tube	< 1 ^e	Sandstone	6	Pells et al. 1980
Rock roller	< 4 ^e	Sandstone	6	Pells et al. 1980
Rock roller	< 1 ^e	Sandstone	30	Pells et al. 1980
Auger	< 10 ^e	Sandstone	6	Pells et al. 1980
Three flight auger	5.7 ^d	Mudstone	0.83	Williams 1980
Three flight auger ^b	12.2 ^d	Mudstone	0.55	Williams 1980
Three flight auger ^b	4.8 ^d	Mudstone	0.59	Williams 1980
Core barrel ^b	10 ^d	Mudstone	0.58	Williams 1980
Bucket auger	7.3 ^d	Mudstone	2.46	Williams 1980
Bucket auger ^b	17.3 ^d	Mudstone	2.30	Williams 1980
Bucket auger	5.9 ^d	Mudstone	2.30	Williams 1980
Bucket auger	5.2 ^d	Mudstone	3.06	Williams 1980
Auger	12.7 ^c	Shale	5.4, 5.6, 11.1	Horvath et al. 1983
Auger + Special pneumatic tool ^b	24.1 ^c	Shale	5.5	Horvath et al. 1983
Auger + Special pneumatic tool ^a	24.1 ^c	Shale	10.4	Horvath et al. 1983
Auger reaming tool ^a	31.5 ^c	Shale	6.75	Horvath et al. 1983
Auger ^b	76 ^c	Sandstone	8.36 – 9.26	Gloss and Briggs 1983
Auger	< 12.5 ^c	Clay shale	0.5 - 5	O'Neil et al. 1996
Auger + grooving teeth tool ^a	11.1 ^d	Siltstone	3.3 – 7.2	Baycan 1996
Auger + grooving teeth tool ^a	8.2 ^d	Siltstone	3.3 – 7.2	Baycan 1996
Downhole jacket + pneumatic grooving tool ^a	100 ^c	Mudstone	3.2	Walter et al. 1997
Downhole jacket + pneumatic grooving tool ^a	100 ^c	Siltstone	8.9	Walter et al. 1997
Downhole jacket + pneumatic grooving tool ^a	100 ^c	Sandstone	11.6	Walter et al. 1997
Auger	10.2 ^d	Sandstone	-	Collingwood (2000)
Flight auger + circumferential groves cut (WRT1) ^a	7.1 ^d	Shale	7.5 – 20	Collingwood (2000)
Flight auger + circumferential groves cut (WRT1) ^a	9 ^d	Shale	1.5 – 7.5	Collingwood (2000)
Flight auger + circumferential groves cut (WRT1) ^a	16.1 ^d	Siltstone	0.8 – 1.35	Collingwood (2000)
Auger	6 ^d	Sandstone	6	Collingwood (2000)
Auger	7 ^d	Conglomerate	6	Collingwood (2000)
Core barrel	2.8 ^d	Basalt	50 – 75	Collingwood (2000)
Core barrel	4.1 ^d	Basalt	75 – 100	Collingwood (2000)
Auger	2.5 ^c	Clay shale	1.2	Nam and Vipulanandan 2008
Core barrel	3.7 ^c	Clay shale	1.2	Nam and Vipulanandan 2008
Auger	1.6 – 2.6 ^c	Clay shale	6	Nam and Vipulanandan 2008
Core barrel	4.2 ^c	Clay shale	2.1	Nam and Vipulanandan 2008
Auger	2.2 ^c	Limestone	10	Nam and Vipulanandan 2008
Core barrel	3.5 ^c	Limestone	10	Nam and Vipulanandan 2008
RCD	1 – 5.1 ^c	Granite	100 – 150	Jeong et al. 2010
RCD	1 – 4 ^c	Gneiss	30 – 130	Jeong et al. 2010
Bit	1 – 7 ^c	Gneiss	5 – 50	Jeong et al. 2010

^a Artificial roughening practices

^b Then roughened (the tool used to increase the roughness is not indicated)

^c Provided in reference of load test data

^d Estimated by the authors using roughness profiles provided in reference of load test data

^e Estimated using roughness classification proposed by Pells et al. (1980), see **Table 9**

Table 8. Empirical factors α and β for average side shear resistance design assuming a relationship of type $f_{ave}[\text{MPa}] = \alpha(\sigma_c[\text{MPa}])^\beta$

Design method	α	β
Rosenberg and Journeaux (1976)	1.11	0.51
Horvath and Kenney (1979)		
- For large piles – diameter ≥ 410 mm		
a) Lower bound	0.20	0.50
b) Upper bound	0.25	0.50
- For small piles – diameter < 410 mm		
a) Lower bound	0.25	0.50
b) Upper bound	0.33	0.50
Meigh and Wolski (1979)	0.22	0.60
Kaderbeck and Reynolds (1981)	0.30	1.00
Williams et al. (1980)	0.44	0.36
Australian Piling Code (1980)	0.15	1.00
Horvath et al. (1983)		
a) Smooth socket (conservative lower value)	0.20	0.50
b) Rough socket (conservative lower value)	0.30	0.50
c) Considering roughness factor, RF	$0.80RF$	0.45
Gupton and Logan (1984)	0.20	1.00
Rowe and Armitage (1987)		
a) For R1, R2, R3 ^b , $b = 0.45$	$0.70b$	0.50
b) For R4 ^b , $b = 0.60$	$0.70b$	0.50
Toh et al. (1989)	0.25	1.00
Hooley and Lefroy (1993)		
a) For $0.25 < \sigma_c$	0.30	1.00
b) For $0.25 \leq \sigma_c \leq 3.00$ MPa	0.15	0.50
c) For $\sigma_c > 3.00$ MPa then $f_{ave,peak} = 0.4$ MPa	-	-
Kulhawy and Phoon (1993)		
a) Upper bound	1.34	0.50
b) Lower bound	0.22	0.50
Carrubba (1997)		
a) Lower bound	0.13	0.50
b) Upper bound	0.25	0.50
Zhang and Einstein (1998)		
a) For R1, R2, R3 ^b	0.40	0.50
b) For R4 ^b	0.80	0.50
O'Neill and Reese (1999) ^a , Canadian Foundation Engineering Manual (2006) ^a and American Association of State Highway and Transportation Officials AASHTO (2008) ^a		
a) Smooth socket	0.20	0.50
b) Rough socket	0.30	0.50
c) Very rough socket (i.e., asperities amplitudes more than 10 mm)	0.60	0.50
Technical Building Code, CTE (2006)	0.20	0.50
Basarkar and Dewaikar (2006) ^a	0.40	0.50
Sagong et al. (2007) ^a	0.38	0.51
Rezazadeh and Eslami (2017) ^a	0.36	0.36
Arioglu et al. (2018) ^a	0.36	0.40

Source: Data from O'Neil et al (1996); Serrano et al. (2008).

^aThis correlation has been added by this paper

^bRoughness classification proposed by Pells et al. (1980), see **Table 9**

Table 9. Roughness classification.

Roughness class	Description
R1	Straight, smooth sided socket, grooves or indentations less than 1-mm deep
R2	Grooves of depth 1-4 mm, width greater than 2-mm, at spacing 50-mm to 200-mm
R3	Grooves of depth 4-10 mm, width greater than 5-mm, at spacing 50-mm to 200-mm
R4	Grooves or undulations of depth > 10-mm, width > 10-mm at spacing 50-mm to 200-mm

Source: Reprinted from *International Journal of Rock Mechanics and Mining Sciences & Geomechanics Abstracts*, Vol. 18(2), P, J. N. Pells, R. K. Rowe, and R. M. Turner, "Experimental investigation into shear for socketed piles in sandstone," pp. 33, © 1981, with permission from Elsevier.

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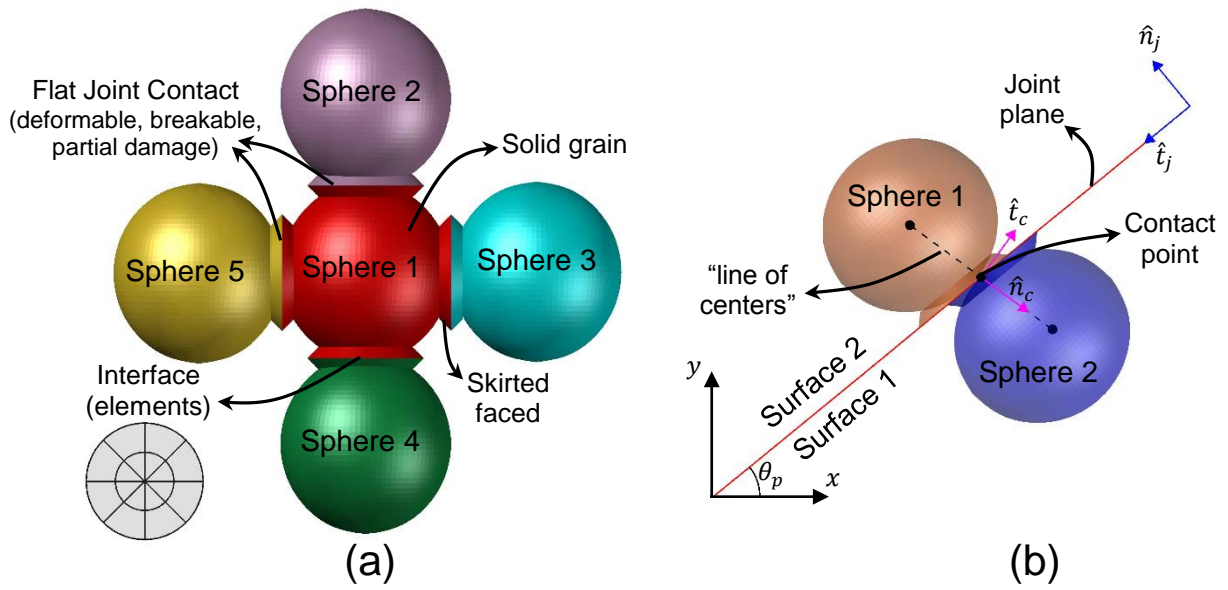


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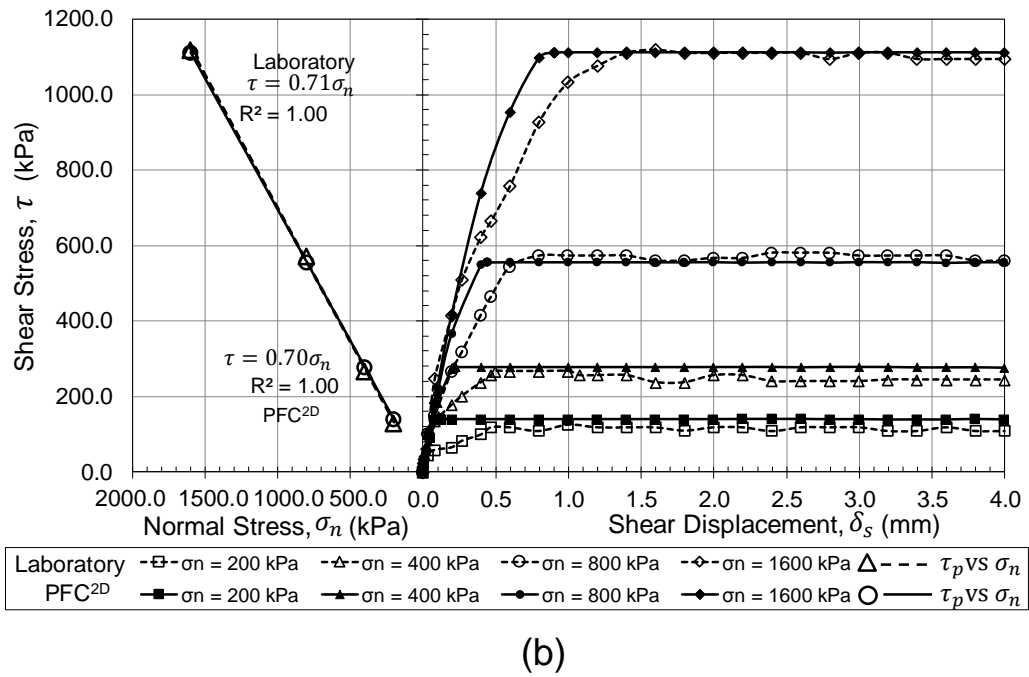
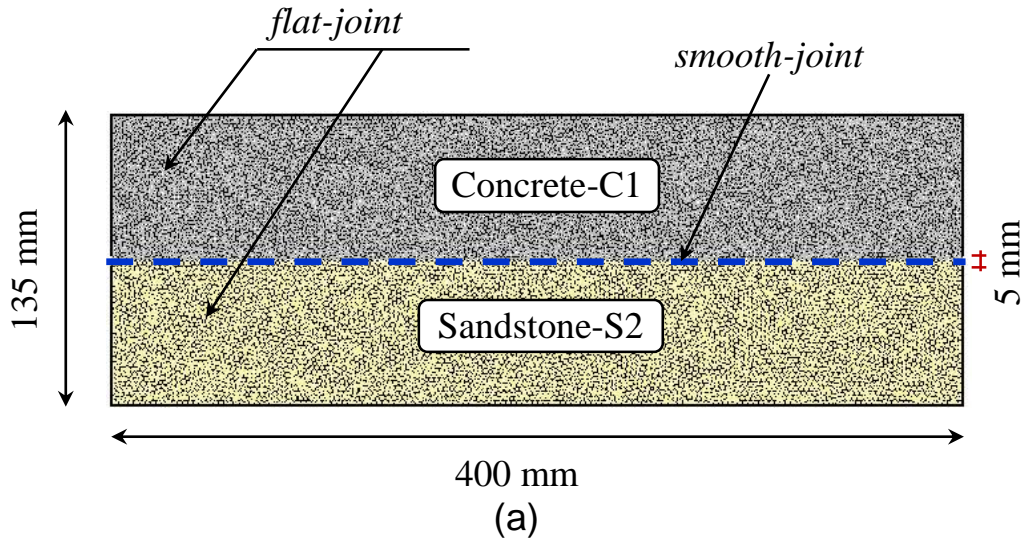


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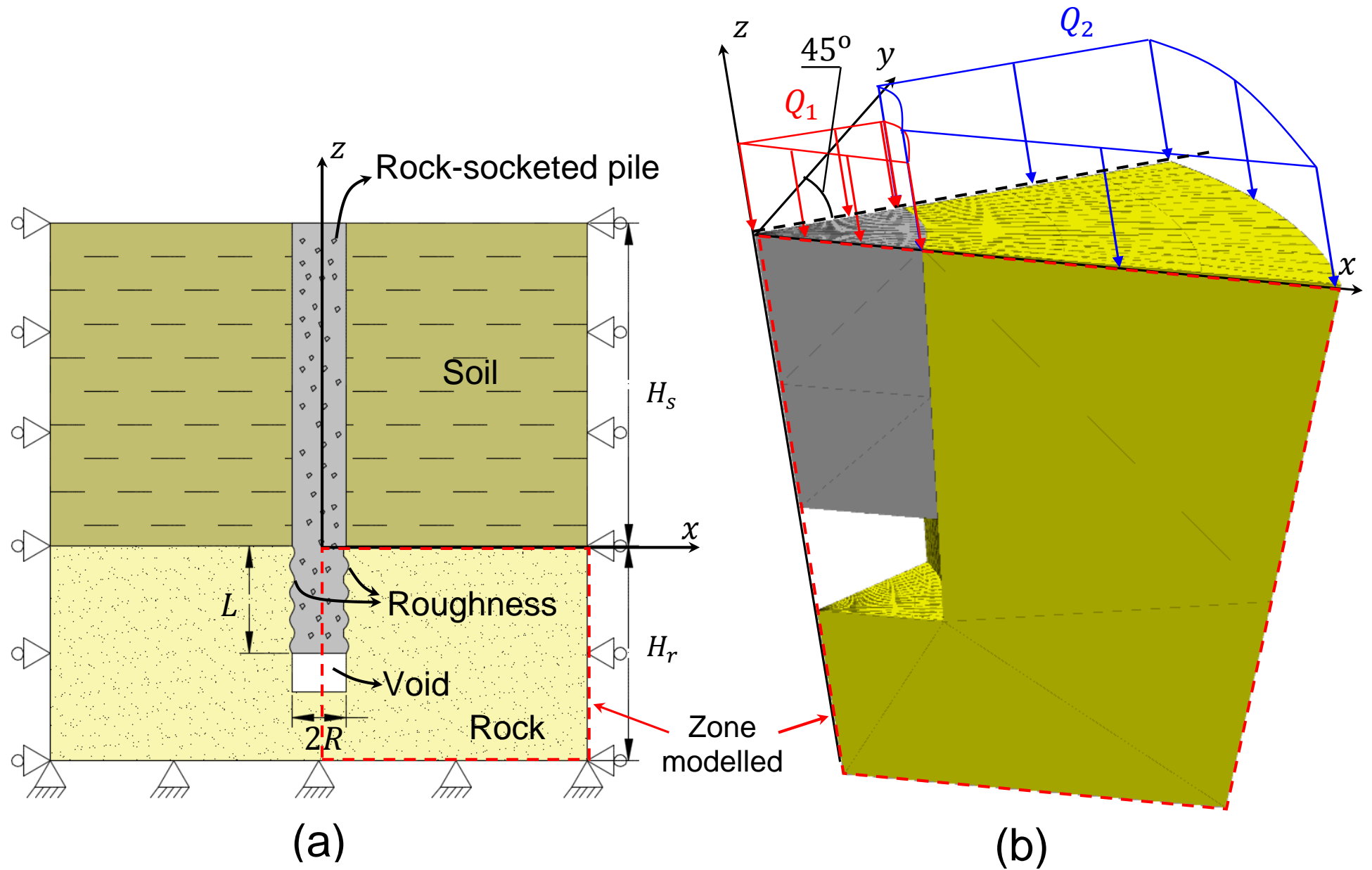


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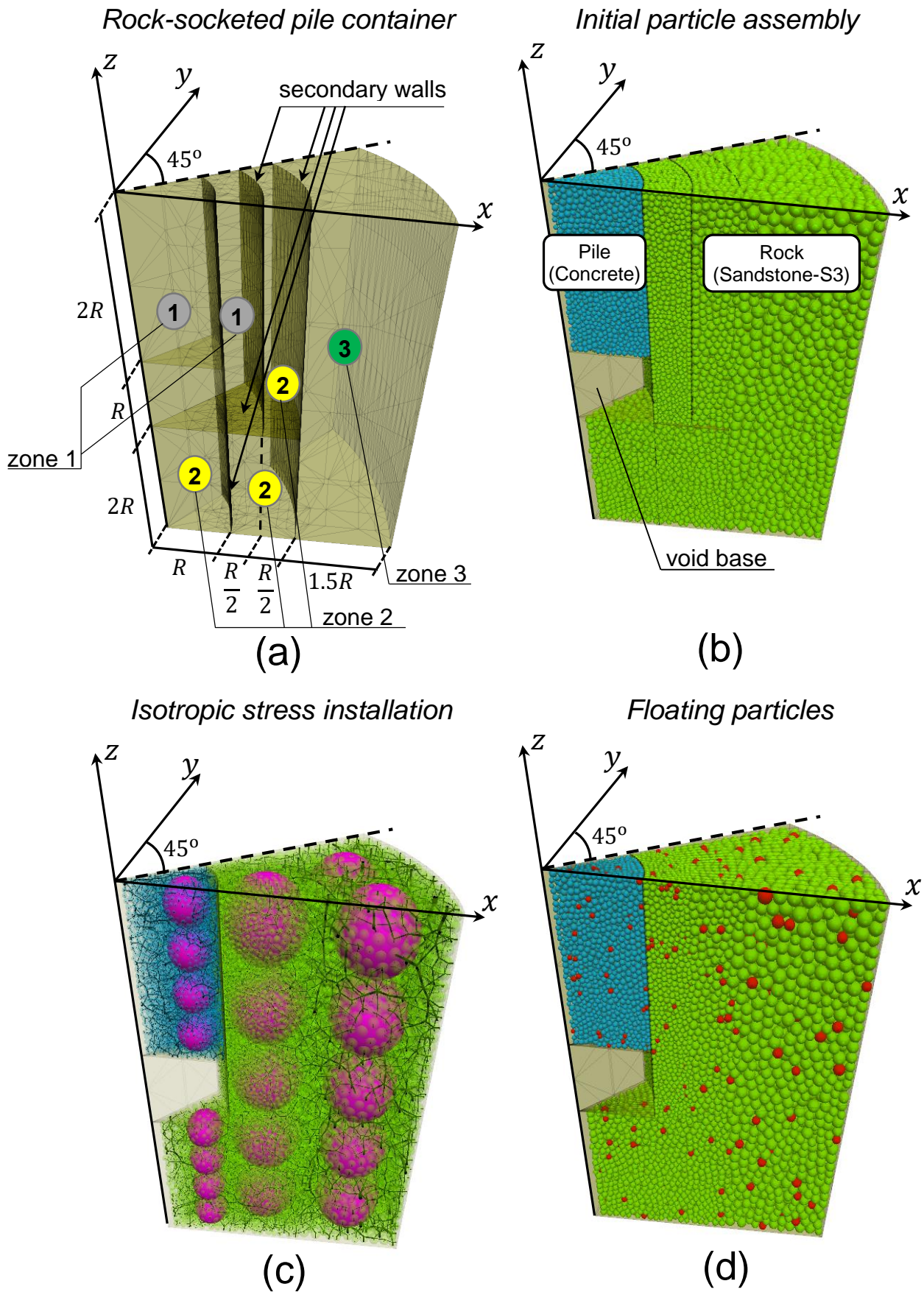
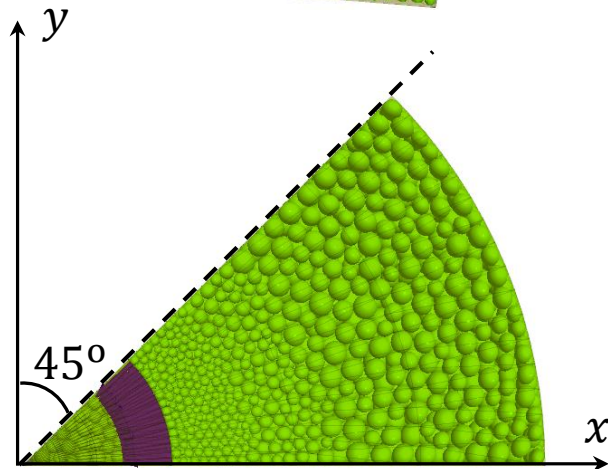
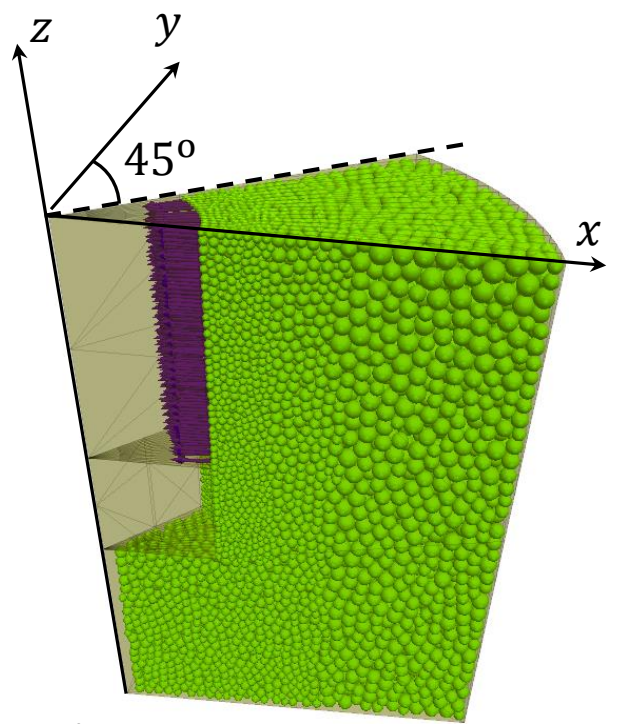
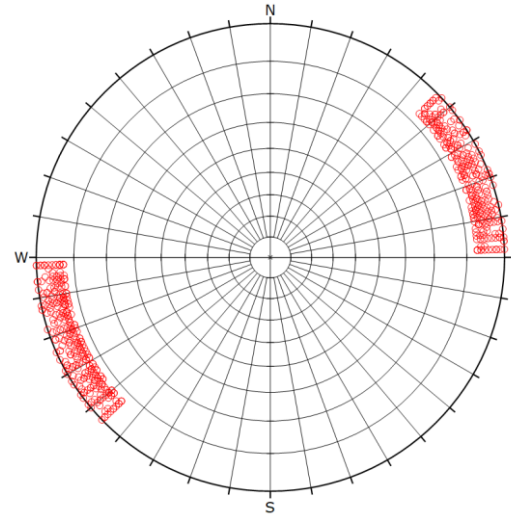


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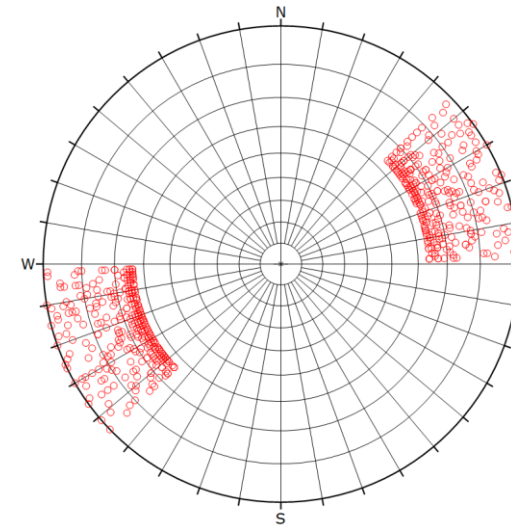
(a)



$RF = 0.025$

Symbol	Feature
○	Pole Vectors

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Projection	Equal Angle



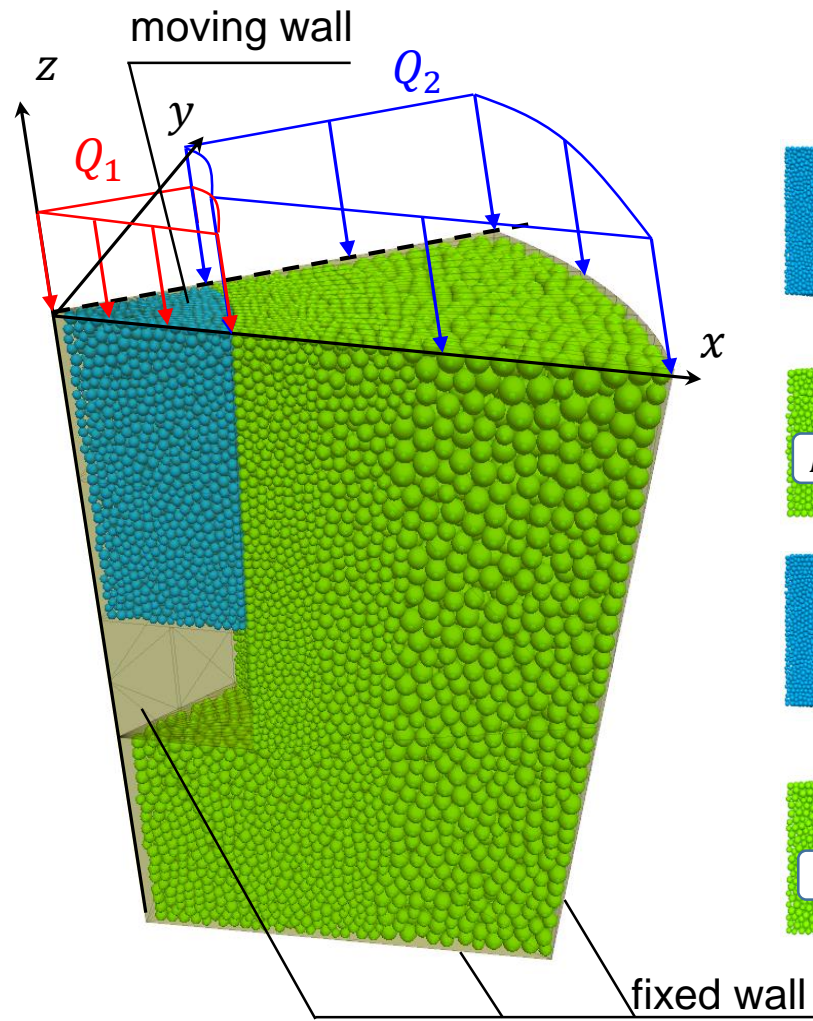
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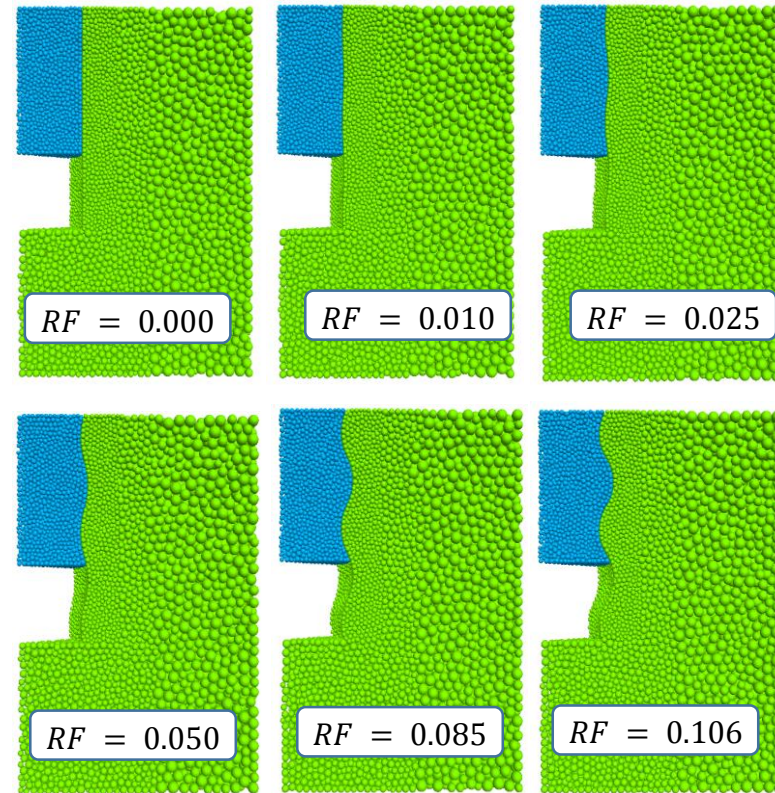
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Vector Count	555 (555 Entries)
Hemisphere	Lower
Projection	Equal Angle

(b)

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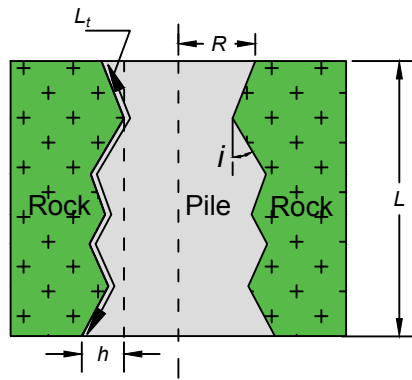


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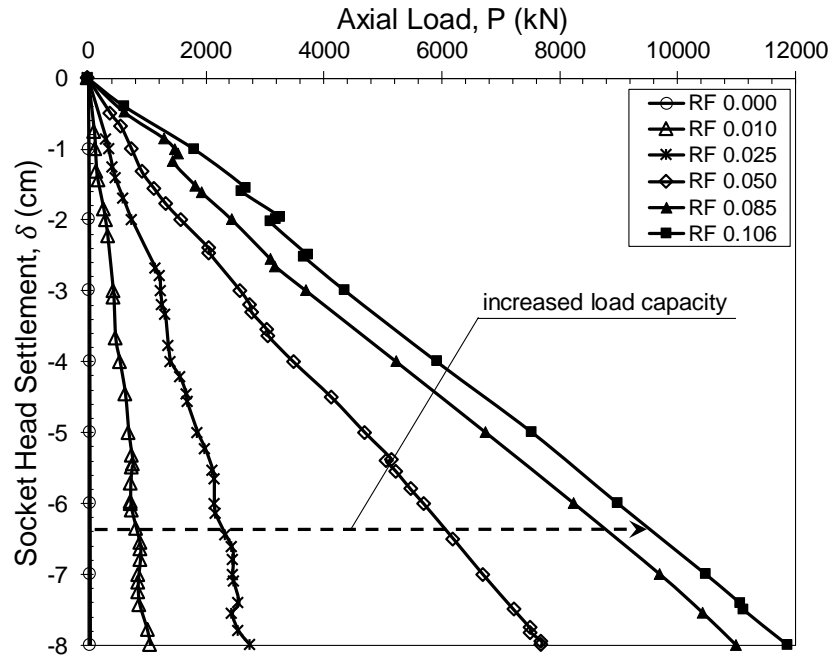


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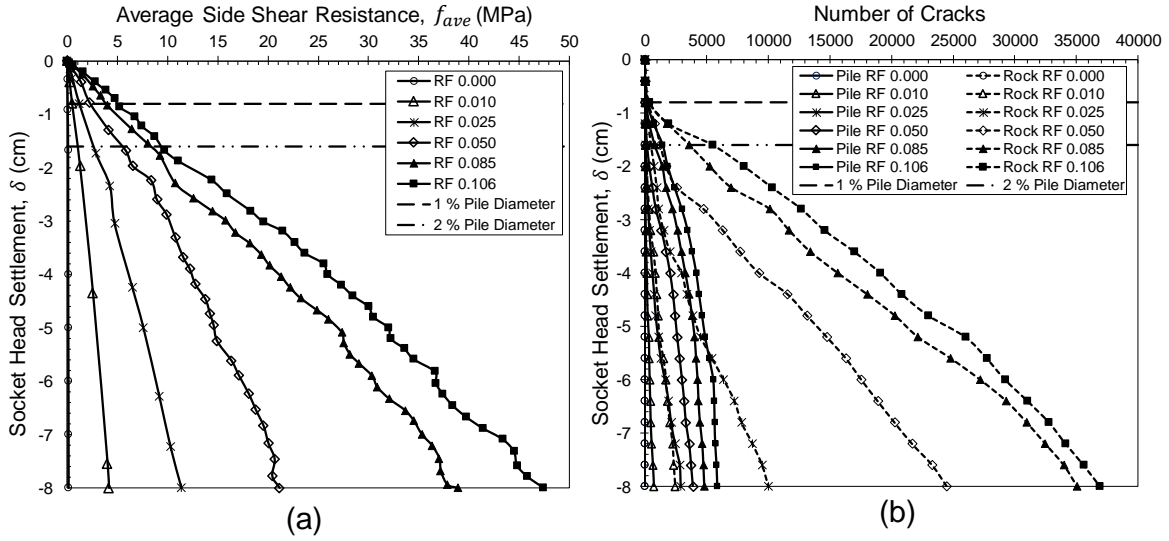


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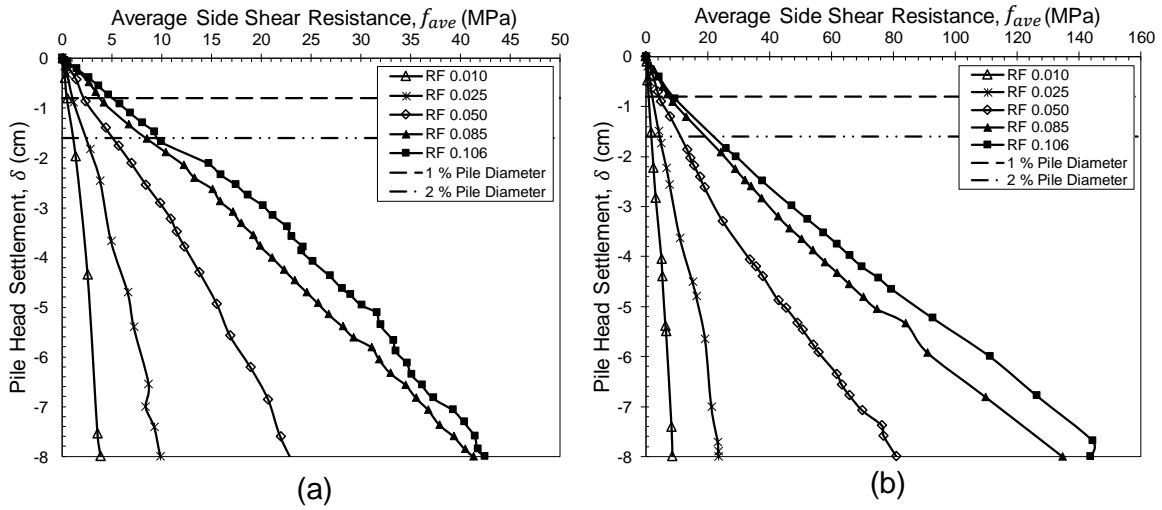


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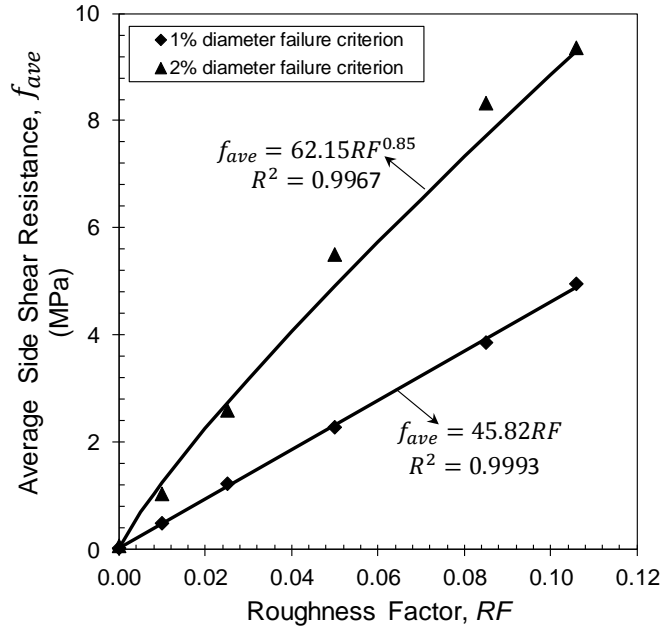


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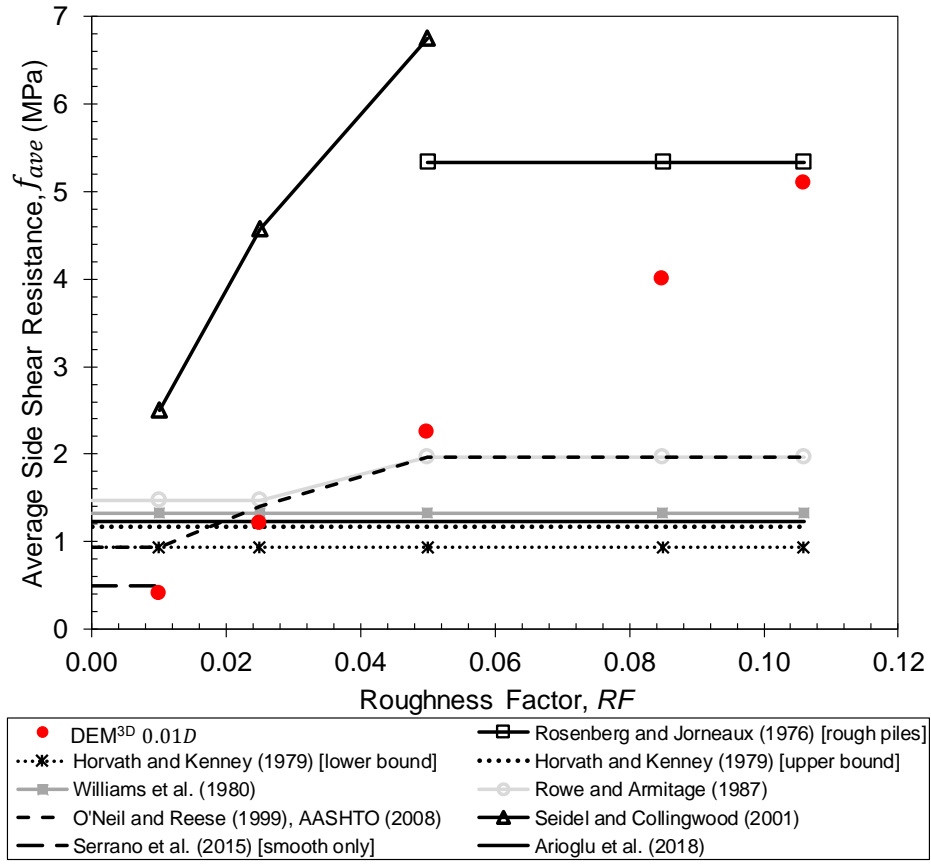


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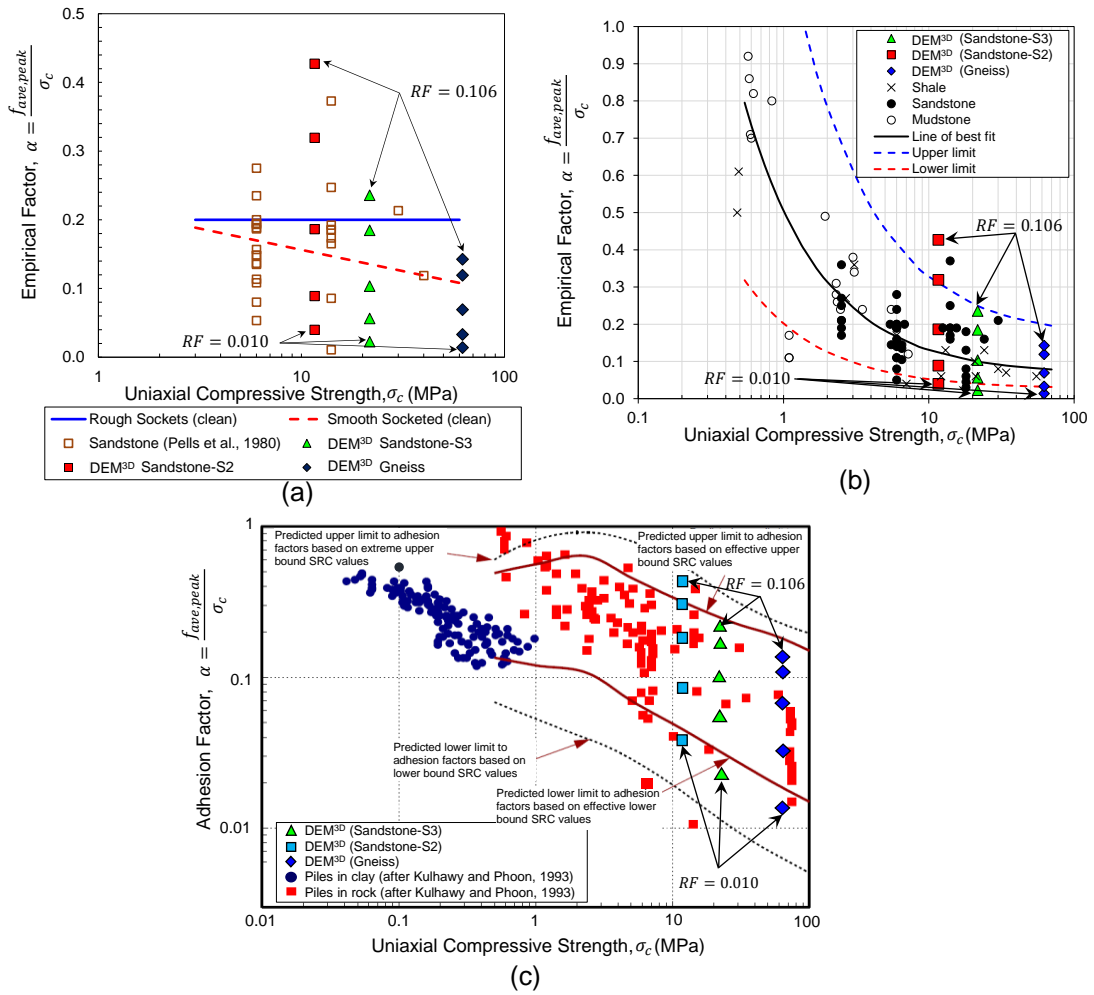


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