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# Effect of Sample Size on the Behavior of Granular Materials

**ABSTRACT:** Natural soils or coarse granular materials containing large particles such as rockfill or ballast are often an issue for geotechnical projects. However, since it is usually complicated or impossible to set up a big enough experimental apparatus to test these materials, their mechanical behavior remains poorly known. To circumvent this difficulty, we have developed a set of triaxial cells of various sizes to study the experimental behavior of granular soils containing particle sizes of up to 160 mm. This paper presents the first results of a two-part experimental study. The first part consists of triaxial tests on sand samples of different sizes in order to examine specimen scale effects. Pre-peak behavior is not affected by the specimen size, whereas post-peak behavior depends on the test conditions that control the development of strain localizations. In the second part, we present drained triaxial compression tests on large samples of calcareous rockfill materials that have parallel grain size distributions. The results show a nonstandard evolution of the friction angle in relationship to grain size if compared to similar studies on coarse granular materials. Although grain crushing occurs during compression, the strain-stress curves of the different materials with different grain sizes do remain close. We explain this phenomenon by the strength evolution of the individual grains with respect to their sizes.

KEYWORDS: triaxial cell, granular materials, rockfill behavior, coarse soil, size effect

#### Introduction

The study of granular materials containing large particles such as rockfills, ballasts, etc, requires suitable laboratory facilities whose dimensions must be in proportion to the maximum size of the grains. In order to satisfy the concept of Representative Elementary Volume, a current assumption in the geotechnical community is to test specimens six to ten times bigger than the particle maximum size (Holtz and Gibbs 1956). Under this condition, the physical and mechanical properties do not depend on the specimen size. Besides, the sample height is usually twice its diameter in triaxial tests in order to lessen the effects of cap and base friction (Bishop and Green 1965; Bishop 1971).

In recent years, the construction all over the world of new and sometimes very high rockfill dams (200 m and more) using natural and local fill materials corresponds to the need for sustainable energy and water resources. The optimum design of such dams involves a better understanding of the behavior of the used materials in order to guarantee the safety of its structural behavior during its

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<sup>4</sup>Département Mesure Auscultation et Calcul Scientifique (MACS), Institut Français des Sciences et Technologies des Transports, de l'Aménagement et des Réseaux, 58 boulevard Lefebvre, Paris 75732 cedex 15, France. lifetime. Other engineering issues (stone columns, ballasts, stone beds above rigid inclusions, etc.) justify the creation of new research programs on the currently ill-determined behavior of coarse granular materials.

The pioneering work on the behavior of coarse soils was done at the University of California, Berkeley (Marachi et al. 1969; Marachi et al. 1972; Becker et al. 1972) and at the University of Mexico (Marsal et al. 1965; Marsal 1967; Marsal et al. 1967) in the 1960s and 1970s. The construction of large facilities then accompanied the scientific objective to extrapolate the behavior of natural rockfills used in dam construction from triaxial tests on finer fractions of the same material. Those experimental programs clearly highlighted the effect of grain crushing on the value of the shear resistance and more particularly on the peak friction angle (Marsal 1967; Marachi et al. 1969; Leps 1970; Marachi et al. 1972; Becker et al. 1972). Extensive triaxial testing programs were completed with either one dimensional compression tests (Marsal 1967) or plane strain compression tests (Marsal et al. 1967; Becker et al. 1972) on the same rockfill granular materials, and all of the test results led to the conclusion that (1) grain breakage is enhanced when stress level, grain size, and sample dimension are increased; and (2) the peak friction angle value decreases if the amount of grain crushing increases, accompanying a more compressible behavior. Such large experimental facilities are still very scarce around the world, although some new devices, among which is the cell described in this paper and others (Verdugo and de la Hoz 2006), attempt to reproduce and extend the previous research programs in light of the present knowledge and technology. Nevertheless, some smaller but still unusual cells have provided interesting results about the behavior of unsaturated gravels and rockfill materials (Chávez and Alonso 2003; Chávez 2004), about the dynamic behavior of ballasts (Indraratna et al. 1998; Lackenby et al. 2007), or about the pseudo-static behavior of rockfills (Charles and Watts 1980; Barton and Kjaernsli 1981; Indraratna et al. 1993; Varadarajan et al. 2006).

In this paper, we present the results of an experimental study based on the development of a set of triaxial cells with different



FIG. 1—Design draughts of the two configurations of the large triaxial cell.

dimensions, which allowed us to address the issue of the maximum grain size effect on the behavior of rockfill material. This set of triaxial cells comprises a conventional triaxial cell for samples of 70 or 100 mm in diameter, a second cell allowing testing samples of 250 mm in diameter, and a very large triaxial cell for samples with a diameter between 500 and 1000 mm.

#### **Experimental Devices**

The small and intermediate triaxial cells, designed to accommodate samples whose diameter is either 70 or 100 mm for the small one or 250 mm for the intermediate one, are not so different from many commercial devices, so that they are not described here. Experiments on relatively coarse soils merely imply that the testing procedure must be adapted as presented below. The description of the very large triaxial cell will be emphasized. Contrary to the peculiar design of Marsal's spherical cell, the principle of the new large apparatus resembles a smaller conventional cylindrical triaxial cell similar to the large test facility at the University of California, Berkeley.

Initially, the cell was designed to support vertical loading up to 1000 kN (first configuration). All the test results on sand samples presented herein refer to this first configuration. Some elements (base of the cell, sample platens, top cover, pre-tension tie rods, piston, and ball joint) were subsequently reinforced to sustain vertical loading up to 2000 kN. The driving hydraulic and electronic system was also simultaneously upgraded; this second configuration was used to carry out triaxial tests on the rockfill material presented in the last section. General sketches of the two configurations are shown in Fig. 1.

The confining chamber (Fig. 2) stands on a pre-stressed concrete slab that can support its total weight (about 150 kN) without any significant deformation. It is placed under a 100 kN travelling crane high enough to pass over the 4.5 m high triaxial device. The cell is composed of a very stiff and fixed steel base  $1.60 \times 1.60$  m large, on which all the cell elements are assembled.

Depending on the sample size and nature, a suitable inner load cell (among three submersible load cells of respective capacities 500, 1000, or 2000 kN and for all of them a linearity lower than  $\pm 0.5$  % of the measure scale) is chosen and installed on the cell base, just below the sample bottom platen directly screwed on the load cell. The use of an inner load cell gives a more accurate measurement of the shear force applied to the sample; it particularly does not integrate the frictional force between the cell cap or the cover plate and the loading piston.

At the present time, the cell may host samples either 1 m in diameter and 1.5 m high (height-to-diameter ratio: 1.5) or 0.5 m in diameter and 1 m high (height-to-diameter ratio: 2). Very stiff distance-pieces are used to adjust the position of the sample inside the cell according to its original height. Of course, the different slenderness of the largest samples imposed by the available total height within the cell may have an effect on the stress-strain responses of the tested material. It has been largely shown that the peak friction angle and the tangent initial stiffness are greater in case the slenderness is reduced or no lubrication techniques are



FIG. 2-Details of the cell elements.

used (Lam and Tatsuoka 1988; Darve et al. 1995; El Bied 2000) because shear stresses are generated due to friction between the platens and the soil sample. Such an effect must be taken into account when comparing the subsequent test results on sand. On the contrary, comparative triaxial tests on rockfill samples were conducted on samples with the same slenderness and the comparison is therefore assumed to be straightforward.

The bottom and top platens are made of steel circular elements 120 mm thick and 1100 mm in diameter for samples 1 m in diameter, and 100 mm thick and 600 mm in diameter for samples 0.5 m in diameter. The inside faces of the platens contain concentric grooves in order to allow drainage. They were also fitted with a metallic screen protected by two fine layers of a textile geomembrane that act as a porous stone and a filter in order to avoid any intrusion of fine particles into the drainage network. No lubrication is used to prevent boundary effects on the platens; experimental observations, however, show that sliding along the end platen is not prevented.

The lateral cylindrical envelope, commonly transparent in most standard triaxial cells, is here replaced by a 2 m high steel doublewall chamber and two heightening elements on both sides of this chamber (Fig. 2). All electrical cables for transducers or nylon hydraulic hosepipes pass through those heightening elements.

The double-wall chamber is composed of

- a steel outer cylinder 12 mm thick and 1256 mm in diameter and
- a high density polyethylene (HDPE) inner cylinder 8 mm thick and 1200 mm in diameter clamped at its tips.

The outer and inner volumes are both filled simultaneously with water and can withstand a confining pressure of 1.5 MPa (it is practically limited to a serviceability pressure of 1 MPa). The driving system continuously controls the equality of the pressures in the two chambers so that the inner HDPE cylinder remains undeformed during the test. Therefore, the inner compartment containing the sample can be assumed to be at constant volume during the test, which enhances the accuracy of the volume change measurements. Two 1 MPa pressure transducers located at the base of the double-wall chamber are connected to the inner and outer volumes, respectively. In case of a difference of pressure on both sides of the HDPE envelope, a servo-valve is automatically opened by the driving system to balance the pressures.

Water volume changes are detected by an electromagnetic flowmeter with an accuracy of  $\pm 0.2$  %, coupled to an electronic counter inserted in the cell pressure line. During shearing at constant cell pressure, the water volume changes are mainly due to the penetration of the loading ram into the inner volume, to the water compressibility, to the membrane penetration effects, and to the sample volume changes (Georgopoulos and Vardoulakis 2005).

Although assumed to be an incompressible fluid, water contracts or dilates whenever a pressure or temperature variation is applied. At temperatures of 0 and 20 °C, the coefficient of compressibility of water  $c_w$  is, respectively,  $5.01 \times 10^{-7}$  kPa<sup>-1</sup> and  $4.57 \times 10^{-7}$  kPa<sup>-1</sup>. The water volume change can be evaluated as (Georgopoulos and Vardoulakis 2005)

$$\Delta V_w = c_w \cdot \Delta \sigma_c \cdot V_w \tag{1}$$

where:

 $\Delta \sigma_c$  = variation in the cell pressure and

 $V_w$ =current water volume in the cell (1.24 m<sup>3</sup> for a 1 m large sample and 2.35 m<sup>3</sup> for a 0.5 m large sample).

A variation in the cell pressure value of 2 kPa (because of a de-

layed response of the hydraulic control system for instance) induces a water volume change of  $1.13 \text{ cm}^3$  for a 1 m sample or  $2.15 \text{ cm}^3$  for a 0.5 m sample at a temperature of  $20^{\circ}$ C, i.e., a sample volumetric strain of about  $10^{-5}$  in the extreme case. Volume changes due to a variation of several Celsius degrees are of the same order of magnitude.

The membrane penetration effects can be assessed following two different approaches: First, by assuming that large samples of coarse soils represent a rigorously homothetic picture of small samples with small particles, the approach by Baldi and Nova (1984) based on triaxial tests on fine sands ( $d_{50} \le 0.5$  mm) can be considered. According to their study, the volume change due to membrane penetration, which mainly occurs during the drained consolidation phase, depends on the mean diameter  $d_{50}$  of the grains, of the sample size (diameter  $d_{sample}=1000$  mm and volume  $V_{sample}$ ), of the effective cell pressure  $\sigma'_c$ , and of the membrane properties (Young's modulus  $E_{membrane} \sim 10$  MPa; thickness  $e_{membrane}=1.4$  mm) and can be expressed by the following relation

$$V_m = \frac{d_{50}}{2 \cdot d_{\text{sample}}} \cdot V_{\text{sample}} \cdot \sqrt[3]{\frac{\sigma'_c \cdot d_{50}}{E_{\text{membrane}} \cdot e_{\text{membrane}}}}$$
(2)

An alternative approach was developed by Frydman et al. (1973), Ramana and Raju (1982), and Nicholson et al. (1989). They suggested that the volumetric membrane compliance per unit membrane area is proportional to the log cycle of the effective confining pressure by a constant *S* called normalized membrane penetration. Various correlations have been established between *S* and a characteristic grain size, either  $d_{50}$  or  $d_{20}$ . Based on a large collection of test results, Nicholson et al. (1989) proposed a relationship between *S* and  $d_{20}$  that tends towards a limiting value of *S* =0.4 (cm<sup>3</sup>/cm<sup>2</sup>)/ $\Delta \log \sigma'_{3}$  when  $d_{20}$  increases up to values higher than 50 mm. This value has been used to correct the initial void ratios of the 1000 mm samples. For the 250 mm samples, according to the relationship proposed by Nicholson et al. (1989), a value of  $S=0.1 (cm^{3}/cm^{2})/\Delta \log \sigma'_{3}$  has been assumed. No correction has been applied in the case of sand samples.

The approach of Baldi and Nova (1984) is suitable for small samples (70 mm) but leads to an overestimated value of the membrane correction (some percent for the volumetric strain) inconsistent with the experimental observations of the volumetric contractancy during isotropic compression for large samples. The second approach considering a value of S=0.1 or 0.4 (cm<sup>3</sup>/cm<sup>2</sup>)/\Delta log  $\sigma'_3$  gives a more realistic value of the volumetric strain during consolidation, close to 1 %, and is therefore used for samples of 250 and 1000 mm in diameter.

Let us now consider a drained shearing phase of a coarse sample at a constant confining pressure. Volume changes due to membrane penetration effects are assumed to be negligible even if the key parameter, namely the characteristic grain size, evolves slightly due to grain crushing. Therefore, the sample volume changes are calculated in the following way

$$\Delta V_{\text{sample}} = \Delta V_{FM} - \Delta V_{\text{piston}}$$
(3)

where:

 $\Delta V_{FM}$ =water volume–entering or exiting the cell–measured by the flow meter installed in the console,

 $\Delta V_{\text{sample}}$  = sample volume change, and

 $\Delta V_{\text{piston}}$ =variation of the cell water volume corresponding to the penetration of the loading piston into the cell ( $\Delta V_{\text{piston}} = \pi . \Delta H . r^2$  where  $\Delta H$  is the relative movement of the loading piston within the

cell determined at the hydraulic jack level, r=40 mm is the piston radius in the first cell configuration, and r=50 mm in the second one).

12 steel tie rods 60 and 1560 mm in diameter and height, respectively, are pre-stressed at a value of 10 MPa to ensure the complete water tightness of the cell once it is finally mounted. They also facilitate the placement of the cover plate by guiding it. The steel circular cover plate, 145 mm thick and 1600 mm large, which closes the cell (Fig. 2), is fitted with a vent valve so that the air can flow out the chamber when it is filled by water and also supports the loading frame, as shown in Fig. 1.

The first configuration of the cell uses a one-way hydraulic jack whose capacity is 1000 kN and displacement course 250 mm. More recently, a new hydraulic jack was acquired in order to apply loading up to 2000 kN statically (from 0 to 50 mm/h) and cyclically with frequencies in the range 0–5 Hz for a course of 300 mm. The hydraulic jack applies a vertical load (in compression or in extension) on the loading steel piston, which slides through the cover plate. The lower tip of the loading piston consists of a ball joint (Fig. 2) attached to the top platen of the sample. Note that the vertical shortening of the sample is measured by an external LVDT bound to the hydraulic jack. The maximum axial strain is limited to 13 % using the 1000 kN configuration of the cell and to 20 % with the 2000 kN configuration.

A critical issue involves the manufacture of the membranes. In order to prevent prohibitive costs and sudden leakage due to membrane tears, a 1.5 mm thick and nonreinforced PVC membrane, usually used as a swimming-pool liner, was chosen. A perfectly rectangular piece slightly greater than the necessary size is carefully cut. For a sample 1 and 1.5 m in diameter and height, respectively, the rectangular piece is 2 m high and 3.19 m long. The ends of the membrane are then thermally glued using a heating tool at a temperature of about 200°C. The membrane, which can be slightly warmed up in order to facilitate its installation, is finally placed on a four-quarter aluminum cylindrical mold and the whole set up is installed on the bottom platen of the cell. Steel rings encircle the membrane on the platens beforehand coated with silicone grease. Because the grain angularity may cause membrane tear, a second PVC membrane and a 0.5 mm thick layer of geo-membrane whose dimensions coincide with the sample dimensions are usually superimposed when coarse soils are tested; the inner membrane is free to slide against the outer membrane, itself fixed on the sample pedestals. The cumulative thickness of the protecting membrane is thus about 3 mm, which is relatively small compared to prior experiments by Marsal (1967) or Marachi et al. (1969).

Before filling the mold, the membrane is pushed against the mold by a vacuum of 50 kPa by using a vacuum pump and a vacuum transducer.

#### **Experiments on Sand Samples**

In order to evaluate the performance of the cell, to improve the experimental procedure and to prove that no material size effects can subsequently affect test results on coarse soils, a series of drained consolidated triaxial compression tests using the 1000 kN configuration of the cell was conducted on dry Loire River sand. It has been previously studied in the laboratory (Dano et al. 2003) and huge and uniform quantities are available. A similar approach preliminary to extensive programs on coarse materials was previously carried out by Marachi et al. (1969) and Becker et al. (1972), for instance. The Loire River sand consists of recent alluvial deposits of the Loire River, close to its mouth in the west of France. It is composed of sub-rounded silica particles and elongated shell fragments. The mean grain size  $d_{50}$  is 520  $\mu$ m. The coefficient of uniformity  $C_u$  is equal to 3.5. Therefore, Loire River sand is poorly graded sand (ASTM D2487-93). The specific weight  $\gamma_s$  of the grains is 26.2 kN/m<sup>3</sup> (ASTM D854-92). The minimum unit weight  $\gamma_{max}$  (ASTM D4254-91) and the maximum unit weight  $\gamma_{max}$  (ASTM D4253-93) are equal to 15.1 kN/m<sup>3</sup> ( $e_{max}$ =0.735) and 18.5 kN/m<sup>3</sup> ( $e_{min}$ =0.416), respectively.

Triaxial tests on samples 100, 500, and 1000 mm in diameter were carried out with nearly the same dry unit weight  $\gamma$ =17.10 kN/m<sup>3</sup>. Tests on 100 mm samples were carried out using a 50 kN loading device. It is worth noting that calibration tests in the largest samples, in which small aluminum boxes were randomly placed before filling, show a scattering of the unit weight of about 0.3  $kN/m^3$  from one point to another, in spite of all the precautions taken to obtain a homogeneous sample. Sample reconstitution for all the configurations follows a quite similar procedure to achieve the same density state. A given mass of sand (0.6, 40, and 110 kg for the 100, 500, and 1000 mm samples, respectively) is poured into the mold then laid out and finally compacted by using a rammer. The process is repeated nine, eight, and nine times for the 100, 500, and 1000 mm samples, respectively. Such a process involves a clear anisotropy, particularly apparent in the largest sample, even through the PVC membrane. The top surface is carefully leveled and protected by a double layer of geo-membrane. The top platen is finally installed and the membrane fixed on it. A vacuum of maximum value 80 kPa is then applied inside the sample whereas the vacuum between the mold and the membrane is simultaneously released. The mold is then removed after slightly lifting the top platen. The appearance of the samples is shown in Fig. 3. The sample is of a cylindrical shape. Moreover, due to the relatively small grain size, surface defaults caused by membrane penetration are negligible.

The cell is then completely mounted and filled with water by simultaneously controlling the level of water on both sides of the inner HDPE envelope. At the end of the filling phase, the cell pressure  $\sigma_c$  is gradually increased up to 100 kPa by using a pump, whereas the vacuum  $u_a$  inside the sample is simultaneously released so that the sum  $(\sigma_c + u_a)$  remains always lower than the test confining effective pressure. At this stage, fully drained conditions are imposed. The cell pressure is finally increased up to the test confining effective pressure. Note that the confining pressures given in this paper are determined at the pressure transducer level, 1 m below the mid height of the samples. The shearing phase is conducted at a velocity of 25 mm/h (1.66 %/h for a sample 1500 mm high; 2.5 %/h for a sample 1000 mm high) at a constant cell pressure (among the following values: 100, 200, 300, and 350 kPa). Repeatability is assessed by carrying out two tests in the same conditions on samples 0.5 m in diameter at a confining pressure of 300 kPa. As the jack is a one-way device, the final unloading phase cannot be recorded. Therefore, at the end of the shearing phase, a vacuum is applied inside the sample, the hydraulic jack is stopped, and the cell pressure is slowly decreased.

The change of the cross section area of the specimen cannot be neglected during the triaxial test. The following formula (NF P 94-074 10/94) has been used for the correction of the cross section area



FIG. 3—Localization patterns after shearing at a cell pressure 200 kPa.

$$S = \frac{V_0 - \Delta V_s - \Delta V_i}{H_0 - \Delta H_s - \Delta H_i} \times \left[1 + \frac{\Delta H_i}{2(H_0 - \Delta H_s)}\right]$$
(4)

where:

 $V_0$ ,  $H_0$ =volume and height of the specimen under the initial vacuum respectively,

 $\Delta V_i$  and  $\Delta H_i$ =change of volume and height during the shearing phase, respectively, and

 $\Delta V_s$  and  $\Delta H_s$ =change of volume and height of the specimen during the consolidation.

Because the consolidation of dry granular material can induce very small volume changes, the variations  $\Delta V_s$  and  $\Delta H_s$  are neglected in the calculation.

The triaxial test results are presented in Fig. 4 for each of the four confining pressures. The test result on the 1000 mm sample and at a confining pressure of 100 kPa is not relevant because of a poor control of the cell pressure, due to inappropriate proportional-integral-derivative regulation factors.

As expected (Habib 1972) and also previously shown by Marachi et al. (1969) on Monterey  $\neq 20$  sand, because the tests are conducted on representative volumes, the soil response does not significantly change with the sample size in the pre-peak domain. No grain breakage is involved in the present case; the grain size distributions were measured before and after the triaxial tests on the 0.5 and 1 m large sand samples. Only a representative mass of sand was sampled in different points of the specimens. The results are presented in Fig. 5 for the 1000 mm samples. Changes in the gradation are not significant and probably due to the sampling itself, even if some random breakages were observed on elongated shell fragments. However, the post-peak behavior exhibits clearly a dependence on the sample size. Only tests conducted at a cell pressure of 200 kPa are analyzed for this purpose. The rate of softening increases with the sample size. However, it is worth reminding that the height-to-diameter ratio is 1.5 for the 1 m sample and 2 for the other sample sizes. Observation of localization patterns reveals that a single shear band appears in the small sample 0.1 m in diameter, whereas in the 1 m sample, a set of parallel and cross shear bands are visible through the membrane, as shown in Fig. 3.

Figure 6 presents the evolution of the ratio  $\eta_{max} = (q/p')_{max}$ where q is the deviatoric stress and p' is the mean effective stress at peak, respectively. The variation of  $\eta_{max}$  for the 350 kPa cell pressure, for instance, corresponds to a variation in the value of the peak friction angle of 2°. In a first approximation, one can observe a decrease of the maximum stress ratio with the stress level, which indicates a curvature of the maximum shear strength envelope. However, the test results are quite inconsistent since samples with a slenderness of 2 show a greater maximum stress ratio and samples with a lower density (500 mm samples) show a bigger maximum stress ratio. The effect of the density heterogeneity and of the strain localization is assumed to explain the observed differences.

The parameters of the linear elastic Mohr–Coulomb perfectly plastic material are given in Tables 1–3, where *E* is the secant elastic modulus at an axial strain of 0.2 %,  $\nu$  is the Poisson's ratio,  $\varphi$  is the peak friction angle determined from the tangent curve to the Mohr's circles, which can be seen as an average peak friction angle,  $\gamma_d$  is the dry unit weight,  $\sigma'_3$  is the confining pressure,  $e_0$  is the void ratio, and  $\psi$  is the dilation angle. The cohesion is null. The peak friction angle value for samples 100 mm in diameter has been validated by another study (Benazzouz 2005), which gives a value of the peak friction angle equal to 39.2° for similar unit weights. The comparison of test data on samples 100 and 500 mm in diameter, for which the slenderness is identical, does not reveal any difference in the peak friction angle. Therefore, we can conclude that there is no obvious sample size effect in the pre-peak behavior.

#### Triaxial Tests on Calcareous Rockfill Samples

Even if the triaxial cell could be used as a calibration chamber, its main objective is to test coarse granular materials. This type of granular soil exhibits peculiar behaviors under compression and shearing that have been related to grain crushing more likely to appear in coarse granular materials. Statistically, large particles can undergo a more intense crushing than smaller particles at stress levels (some hundreds of kilopascals) currently encountered in civil engineering for the same mineralogy. The amount of crushing depends on many key factors such as stress levels, stress paths (compression or shearing), density, moisture (Chávez and Alonso 2003), grain morphology and mineralogy, and grain size distribution (Hardin 1985; Biarez and Hicher 1997). At the grain scale, many fracture patterns can be created depending not only on the individual strength of the grains but also on the coordination number and the way local forces are applied to the grains (Guimaraes et al. 2007). For instance, loose assemblies have smaller coordination numbers; splitting mode fracture is then enhanced. On the contrary, in dense granular assemblies, loads at the particle scale are statistically more uniformly distributed on the grains, which therefore flake (Guimaraes et al. 2007). However, this may not always be the case since it is well known that external loads are mainly transmitted through strong force chains into the granular materials, thus enhancing the splitting mode fractures (See, for instance, Richefeu et al. 2009).

Other parameters can also affect the material behavior. Using the large triaxial facility at Idiem, University of Santiago (Chile), De La Hoz (2007) tested different coarse soils with different break-



FIG. 4-Triaxial tests results on Loire River sand.

age capabilities under drained triaxial conditions. He chose to test parallel grain size distributions of the same mineral stock, as previously done by Marachi et al. (1972). The diameter of the particles reached 100 mm. At a given stress level for some granular materials, the peak friction angle decreased when the mean grain size increased and grain crushing was relatively intense. On the other hand, the peak friction angle would increase when the grain size distributions were not perfectly parallel. The increase was found to be related to the quantity of fine particles in the initial grain size distribution as well as to a dependence of the angularity with the



FIG. 5—Comparison of gradations before and after triaxial tests for sample 1000 mm.



FIG. 6—Evolution of the maximum ratio  $\eta_{max} = (q/p')_{max}$  for sand samples.

mean diameter. An ultimate case for which the peak friction angle as well as the initial tangent stiffness did not depend on the mean grain size was obtained with strictly parallel grain size distributions, a low proportion of fines, the same density (no variation in the minimum and maximum dry densities), a similar angularity of the grains for all dimensions, and mainly the absence of grain crushing. Except the last one, these conditions are those of the subsequent tests on the calcareous rockfills carried out with the three triaxial cells.

In this section, another condition for an almost constant maximum stress ratio at the same stress level and at a same strain level but with different mean grain sizes was obtained. Triaxial test results on samples of a calcareous granular material 100 (Fig. 7(a)), 250 (Fig. 7(b)), and 1000 mm (Fig. 7(c)) in diameter, respectively, are compared. The height-to-diameter ratio is 1.5 for all configurations. The tested material is a calcareous rockfill from a quarry located in Préfontaines (center of France). Initial gradations for the three samples size are parallel, as suggested by Lowe (1964), with a uniformity coefficient  $U_c$  close to 2 (Fig. 8). Grain sizes are in the range (1.25-12.5) mm for the 70 mm samples, (5-40) mm for the 250 mm samples, and (25-160) mm for the 1000 mm samples. The angularity of the particles, classified as sub-angular (ASTM D2488-10), is shown in Fig. 9. The specific unit weight measured on small particles is 26.8 kN/m<sup>3</sup>. The material is first dried by natural evaporation in a covered hall; the water content initially equal to 11 % decreases until a value of 0.3 % is reached. Equilibrium between atmospheric conditions in the hall and within the material is assumed.

Minimum and maximum void ratios cannot be experimentally determined on the coarse fraction because of the grain size. Therefore, a relative density cannot be defined and we chose to conduct the tests at a same density rather than at a same relative density. This choice was also corroborated by the fact that the grains present a similar angularity at each scale. Becker et al. (1972) indicated that the reduction of the friction angle value was lower for tests at a fixed relative index than for tests at the same initial void ratio.

 TABLE 1—Parameters of the linear elastic perfectly plastic model for Loire river sand specimens of 100 mm.

$\sigma'_{3}$	$\gamma_d$ kN/m <sup>3</sup>	0	$E_{\text{sec}} (0.2 \%)$	11	ψ degree	φ degree
100 KI a	17.13	0.53	65	0.322	11.5	39.6
200	17.35	0.51	94	0.273	9.3	
300	17.31	0.51	85	0.258	13.3	
350	17.01	0.54	75	0.25	8.2	

TABLE 2—Parameters of the linear elastic perfectly plastic model for Loire river sand specimens of 500 mm.

σ' <sub>3</sub> kPa	$\frac{\gamma_d}{\mathrm{kN}/\mathrm{m}^3}$	$e_0$	<i>E</i> <sub>sec</sub> (0.2 %) MPa	ν	ψ degree	φ degree
100	17.10	0.53	55	0.10	15.1	39.5
200	17.06	0.54	65	0.22	13.3	
300	17.08	0.53	72	0.30	9.8	
350	17.02	0.54	63	0.21	11.3	

TABLE 3—Parameters of the linear elastic perfectly plastic model for Loire river sand specimens of 1000 mm.

$\sigma'_{3}$	$\gamma_d$		$E_{\rm sec} (0.2 \%)$		ψ	φ
kPa	kN/m <sup>3</sup>	$e_0$	MPa	ν	degree	degree
200	17.09	0.53	63	0.36	8.2	36.9
350	17.00	0.54	70	0.28	8.5	



FIG. 7-Triaxial test results on rockfill specimens.

Test characteristics are indicated in Table 4. The rockfill material for the 1000 mm samples is deposited by 15 or 17 successive layers of 100 kg. Each of them is then compacted using a hand-held vibratory needle inserted at different locations into the rockfill layer during 60 s for the unit weight of 11.9 kN/m<sup>3</sup> (estimated relative density: 52 %) and more than 5 min to obtain a unit weight of about



FIG. 8—Normalized grain size distributions for rockfill triaxial testing.

14.3 kN/m<sup>3</sup> (estimated relative density: 85 %). Note that Test 1000-200B was conducted at a lower unit weight of 14.05 kN/m<sup>3</sup>. The relative densities have been estimated from  $e_{\rm min}$  and  $e_{\rm max}$  measurements on the small grain size fraction. The height of each layer is controlled so that homogeneity through the sample can be assumed. Two layers of PVC membrane and one layer of geomembrane surround the rockfill material. A similar procedure was adopted for the 70 and 250 mm samples. The 70 mm specimens were surrounded by two layers of 0.5 mm latex or 1 mm neoprene membrane while the 250 mm samples, tested under a 500 kN loading system, were surrounded by two 1 mm thick neoprene membranes.

The shearing rate was 1.67 %/h for the loosest sample (using the 1000 kN configuration) and 2 %/h for the densest samples (using the 2000 kN configuration). In order to compare grain size evolutions, the shearing phase was stopped at the same maximum axial strain of 20 % for all the tests carried out on the 2000 kN configu-



FIG. 9—Particles morphology as a function of size.

Name	Diameter <sup>a</sup> (mm)	Height (mm)	Mass (kg)	Unit Weight (kN/m <sup>3</sup> )	Void Ratio <sup>b</sup>
70-100A	66.2	105.0	0.527	14.30	0.86
70-100B	66.0	105.9	0.530	14.35	0.86
70-100C	65.7	105.6	0.529	14.49	0.84
70-200A	65.5	105.9	0.522	14.34	0.85
70-200B	65.9	106.0	0.529	14.37	0.85
70-400A	66.1	106.0	0.544	14.70	0.80
70-400B	65.4	103.8	0.521	14.63	0.80
70-400C	65.8	105.4	0.531	14.55	0.81
250-100A	245.2	379.8	26.61	14.56	0.83
250-100B	244.9	380.1	26.51	14.52	0.84
250-200A	245.9	377.0	26.45	14.49	0.83
250-200B	245.9	381.0	26.40	14.31	0.85
250-200C	245.9	379.0	26.55	14.47	0.83
250-400A	246.0	377.5	26.55	14.51	0.82
250-400B	246.0	382.8	26.51	14.29	0.85
1000-100	1003	1496	1707	14.34	0.86
1000-200A	997	1503	1526	12.91	1.06
1000-200B	1006	1500	1687	14.05	0.89
1000-400	1004	1506	1709	14.23	0.86

TABLE 4—Calcareous samples characteristics for triaxial tests.

<sup>a</sup>Corrected diameter=external diameter minus four times the membrane thickness.

<sup>b</sup>Corrected value including membrane penetration effect based on  $S=0.4 \text{ cm}^3/\text{cm}^2$  per log cycle of  $\sigma'_c$  for 1000 mm samples and  $S=0.03 \text{ cm}^3/\text{cm}^2$  per log cycle of  $\sigma'_c$  for 70 mm samples (Nicholson et al. 1989), assuming no particle crushing, which favours compressibility during consolidation.

ration. At the end of the test, the membrane was cut at its base and gradually opened over a height of about 0.6 m. Contrary to an uncemented sand mass, which would immediately flow, the rockfill mass did not collapse and photos of the final structure could be taken (Fig. 10). The observations show intact grains even of large diameters, some grains deeply crushed, and some grains that split into two parts. It was not possible to distinguish specific zones of strong crushing or zones not affected by grain breakage. In other words, crushing seems to be uniformly distributed within the sample volume. Accumulation of fine particles at the base of the sample was observed and associated with a possible surface degradation of the grains due to frictional attrition.

The deviatoric stress versus axial strain curves and the volumetric strain versus axial strain curves are presented in Fig. 7. The comparison between the 70, 250, and 1000 mm samples does not show any significant change in the shear strength, contrary to prior experimental observations by Marsal (1967) or Marachi et al. (1969). This conclusion cannot be strictly extended to the peak friction angle since the maximum deviator stress was not reached in all of the tests. Comparing the stress ratio  $\eta_{max} = (q/p')_{max}$  at the same axial strain of 20 % shows a decrease of the stress ratio with confining pressure (Fig. 11). At any given confining pressure, the differences in the stress ratio values can be mainly attributed to differences in initial densities, the 250 mm samples being slightly denser than the 70 and 1000 mm ones.

Grain size distributions carried out after the triaxial compressions are presented in Fig. 8. They clearly show that for a given grain size distribution, the amount of grain crushing increases with the confining stress (Fig. 8(b)), and that at a given confining stress, the 1000 mm samples present a slightly greater amount of particle breakage related to a lower maximum stress ratio. Even if nonsignificant, the lower amount of grain crushing for denser samples, namely samples of 250 mm in diameter, can be correlated with greater shear strength.

To understand why no size effect appeared in the present tests, a series of crushing tests on individual particles placed between two stiff plates was carried out, as previously done by Marsal (1967). The average diameter  $d_{ave}$  of the crushed particles was in the range 10-100 mm. All the crushing tests exhibited an elastic-fragile behavior so that it was easy to measure the maximum force  $F_{\text{max}}$  as a function of the average diameter (Fig. 12). The fitting curve gives a power law with a power coefficient of 1.90 (Fig. 12). This means that the failure stress  $\sigma_f = F_{\text{max}}/d_{\text{ave}}^2$  decreases only slightly with  $d_{\text{ave}}$ . Consequently, there is no longer an obvious size effect in particle individual strength. It will be demonstrated in another paper (Frossard et al. 2010; Frossard 1979), from a micro-macro analysis, that the decrease of the maximum stress ratio is directly related to the individual strength properties of the grains. Previous experimental observations by De La Hoz (2007), Marsal et al. (1965), Marsal (1967), Marachi et al. (1969), and Marachi et al. (1972) confirm this finding. Several numerical models (Daouadji et al. 2001; Muir Wood and Maeda 2008) assume a relationship between micromechanical properties, grain crushing, and macroscopic behavior.

#### Conclusions

This paper presents triaxial test results by using a set of triaxial devices among which is a large triaxial cell designed for investigating coarse soil behavior. The main characteristics of the cell as well as the main difficulties in the use of this facility have also been evoked. At least two sample configurations can be tested. The most recent evolutions allow the users to apply confining pressures in the range 100–1000 kPa and axial loading up to 2000 kN. The original features of the cell beyond its dimensions can be attributed to the different possible sample sizes and the double-walled cylindrical





FIG. 12-Individual grain crushing tests.



(b)

FIG. 10-Final granular structure showing grain crushing.

envelope designed to accurately measure the volume changes of the samples during testing.

The first set of tests carried out on sand samples attempt to show that size effects cannot affect the results of drained triaxial compression tests before peak. Test results on samples 100, 500, and 1000 mm in diameter were compared. The pre-peak behavior is not



FIG. 11—Evolution of the maximum ratio  $\eta_{max} = (q/p')_{max}$  for calcareous samples.

significantly dependent on the sample size, as assumed by the Representative Elementary Volume concept. Drained triaxial tests on a calcareous rockfill show that with the large triaxial cell we are capable of carrying out experiments on coarse soils and quantify the amount of grain crushing. In the specific case of the calcareous material, a nonstandard behavior was highlighted: The shear strength at a given axial strain level does not decrease as samples with larger particles are tested. It is the correlation with the individual strength of the grains that explains such a peculiar behavior.

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